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Minto Mine Constructed Wetland Treatment Research Program - Site Assessment

Prepared for

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1. Abbreviations and Definitions

CWTS(s) – Constructed Wetland Treatment System(s)

DNA (deoxyribonucleic acid) – is the hereditary material in humans and almost all other organisms. This can be used to identify organisms.

DO – dissolved oxygen

DSTSF – Dry Stack Tailings Storage Facility

Heterotroph – an organism deriving its nutritional requirements from complex organic substances.

Metagenomics – the study of genetic material recovered directly from environmental samples, related to a community rather than an individual organism.

Microbes – microscopic organisms which may be uni- or multi-cellular. This includes algae, bacteria, fungi, viruses, and yeast.

MPN (Most Probable Number Method) – a statistical value representing the viable population of microbes in a sample through use of dilution and multiple inoculations.

MVFE – Mill Valley Fill Extension

Oxidation – is the loss of electrons or increase in oxidation state by a molecule, atom, or ion. Can be catalyzed by microbes.

Redox – Oxidation-reduction potential, measured in millivolts

Reduction (chemical) – is the gain of electrons or a decrease in oxidation state by a molecule, atom, or ion. Can be catalyzed by microbes.

SeIV broth or agar – laboratory growth medium that contains selenite (SeIV) and indicates reduction of selenite to elemental selenium through a visible colour change.

Species – one of the basic units of biological classification and a taxonomic rank. Rank in the classification of organisms below genus and above strain.

SWD – Southwest Waste Dump

Taxonomy – classification, identification, and naming of organisms.

TSS – total suspended solids

YTS250 (Yeast, Tryptone, Starch, 250g each) – a type of laboratory growth medium used for environmental microbes.

2. Introduction

Constructed Wetland Treatment Systems (CWTSs) have been identified as a contingency method for improvement of the quality of site runoff water in the post-closure period for the Minto Mine (Minto Phase V/VI Expansion Project, YOR Project Number 2013-0100). Once established, wetlands can become self-sustaining ecosystems with the plants providing yearly renewal of carbon to fuel microbial activity. As such, they possess the desirable potential to remediate contaminated mine drainage for as long as it is generated. In order for the CWTS to be effective, they must be designed, piloted, optimized, implemented, and maintained in a site-specific manner.

At the request of Minto Explorations Ltd, Contango Strategies has undertaken a site assessment to inform the design of site-specific passive treatment technologies for the Minto site during post-closure. The overriding goal of the study was to identify natural water treatment processes that are currently occurring at the Minto site in areas receiving impacted waters, and how these could be optimized for passive water treatment in closure. As such, the focus of the research program reported here was on areas that currently receive impacted waters (Figure 1), with emphasis on the W15 monitoring station and upstream creek and wetland area that receive seepage and runoff from the Southwest Waste Dump (SWD). Additionally, preliminary information was gathered for the W37 area receiving seepage and runoff from the Dry Stack Tailings Storage Facility (DSTSF) and Mill Valley Fill Extension (MVFE) toe area. As was outlined in Appendix W – Preliminary Reclamation and Closure Plan Phase V-VI of the Phase V-VI expansion plan, the focus of work at this point is on the W15 area and SWD. At a future date, additional work will occur in the future down gradient of the DSTSF and within the footprint of the current water storage pond. The site assessment included two visits to the Minto Mine to gather information and samples. Sampling consisted of in situ measurements, collection of samples for physiochemical, geochemical, and microbiological analyses. General

feasibility of implementing passive treatment was also evaluated during the site visits.

It is our understanding that before this site assessment of the natural wetland area, there has been no monitoring of water quality along the creek between the W10 and W15 monitoring points, of the creek entering the W15 monitoring point, or the W15 runoff area entering the W15 monitoring point. Therefore, this study provides new information to inform decisions regarding the plausibility of water treatment by constructed wetlands at the Minto Site.

Copper is of greatest interest in this study, however, there are a number of other elements and parameters that can influence the biogeochemistry of a system and overall treatability of the copper. These include the presence of iron and sulphur, as well as the pH and oxygen content of the environment. Additionally, there are a few other elements (e.g., selenium) that will require some treatment in order to achieve discharge goals.

3. Site Visits

3.1. Information Gathering

Two site visits were conducted as part of the site assessment for the CWTS. Prior to the first site visit, preparations included gathering site-specific historical and predicted closure information including:

- Rates of seepage for known seeps (and interpretation as a function of dump area and catchments)
- References to analytical results about the site
- Rate of flow for W15 and W37 based on existing data
- Aerial photo interpretation or site personnel knowledge of areas with potentially acceptable granular soil materials, peat, limestone, and organic soils for subsequent on-site materials investigation, inspection and testing.
- Use of site-wide vegetation map to identify areas for possible vegetation borrow sources for the CWTS.
- Determine water quality parameters to be monitored and concentrations that will be used for pilot tests.

3.2. First Site Visit

The first site visit was conducted mid-August 2013 and had a primary goal of familiarizing Contango personnel with the site layout and water flows in order

to properly design sampling plans for assessment of existing natural treatment mechanisms occurring on site. Specific information gathered in the first site visit included:

- Visual inspection along the toe of SWD, Main Dump, DSTSF, and area down-gradient of W37 to identify and document primary seeps that have the potential to be intercepted along with an assessment of current fate of seepage surface flows.
- Identify conceptual options for intercepting and routing seepage into the CWTS.
- Selection of a significant seep(s) for further test work based on initial site reconnaissance. Evaluate potential path of seepage interception trench in comparison to seep location(s).
- Identify probable locations for CWTS. Investigation and evaluation of natural remediation processes that may be occurring on site as identified by redox and pH changes between surface water and plant roots, and choice of plant types to be tested in Contango's pilot CWTS tests.
- Identification of borrow sites for CWTS substrate (limestone, coarse sand/gravel, peat, plant sources for much, etc.)
- Identify ideal CWTS plant species based on site observations of vegetation.

Information collected prior to and during the first site investigation were reviewed and then used to refine the scope and direction of data collection in the follow up site visit. The refinement of scope and testing locations was conducted in consultation with ACG and staff from the Minto Mine.

3.3. Second Site Visit

A second site visit took place September 20-23, 2013 based on observations from the first site visit and historical and modeling data compiled to date. The purpose of the second site visit was to collect water, soil, plant, and microbiological samples from the natural wetland and creek area flowing between the W10 to W15 monitoring stations. This area begins with a creek that flows from off site through the W10 area and subsequently receives seepage and surface runoff from the SWD, surface runoff from the catchment area (including non-contact waters), and road runoff from a culvert entering the W15 collection pond. Samples were taken from potential borrow sources to assess suitability of the materials for constructing soils for the CWTS. Additionally, *Carex aquatilis* (Water Sedge) and *Sphagnum* (moss) plants were collected at the W10 site and transported back to Contango's CWTS research facilities in Saskatoon for use in the pilot-scale systems to be built specifically for testing and optimization of designs for the Minto site.

4. Experimental Setup

4.1. Sample and Analytical Data Collection

During the second site visit, fourteen different locations were sampled at the Minto Mine Site (Table 1 and Figure 2). Samples were collected starting at the most downstream location (W15) and progressing upstream to the W10 area over a period of 2 days. In all cases, water samples were collected first, then soil and plant samples. Water and soil samples were collected by grab sample, with soil sampled to a maximum depth of 5-7 cm. Water, sediment, and plant samples were submitted to Maxxam Analytics and analyzed for parameters listed in Table 2. For plants, only the above water (or above ground) portion was analyzed. For *Carex aquatilis*, *Equisetum* sp. and *Sphagnum* sp., the entire above ground (or water) portion was analyzed. For *Salix* sp., only the recent year's growth was collected. All micro samples were processed according to Sections 4.2 and 4.3.

4.2. Growth-based analyses (MPN)

The most-probable number (MPN) of bacteria was determined for all micro samples (Table 1). The test allows for estimation of the number of bacteria that can grow in a specific laboratory medium. The mediums tested for in this project were specific for bacteria that can reduce nitrate, selenite, and sulfate, respectively, as well as the total number of heterotrophic bacteria for comparative purposes (i.e., a growth-permissive test for all bacteria capable of growing on organic carbon as an energy source). MPN tests for sulphate (Active Standard ASTM D4412) were performed under anaerobic conditions, whereas selenite (Siddique et al., 2006) and nitrate (Nitrate Reduction Test, supplied by Sigma-Aldrich) were tested under aerobic conditions. Total heterotrophs (grown with YTS250 medium; Lefrançois et al., 2010) were tested for in both aerobic and anaerobic conditions. In brief, sediment samples were diluted 1:10 with a 0.1% peptone solution. This dilution was then used to perform a serial dilution along a sterile 96-microwell round-bottom plate containing the respective growth media. Sediment dilutions tested in the MPN plate ranged from 1/400 to 1/419,430,400. At a minimum, all tests were performed in duplicate. Wells were incubated at +30°C in a dark incubator and assessed for visible growth (formation of a bacterial pellet) and/or colour change specific to the type of media after 16 (aerobic) or 28 (anaerobic) days. A colour change to black or red/orange indicated sulphate or selenite reduction, respectively. Nitrate reduction was tested as per the kit manufacturer's protocol by addition of reagents to differentiate between reduction to nitrite or nitrogen gas. The most probable number of bacteria capable of each metabolism was then calculated as

described in the FDA Bacteriological Analytical Manual, Appendix 2: Most Probable Number from Serial Dilutions: (http://www.fda.gov/Food/FoodScienceResearch/LaboratoryMethods/ucm109656.htm).

4.3. DNA-based analyses

DNA was extracted from all samples using the MO BIO Powersoil Powerlyzer DNA extraction kit. To identify the bacteria present in each sample, we performed targeted sequencing. The gene that is targeted by polymerase chain reaction (PCR) for bacteria is the v4 region of the 16S ribosomal RNA gene (Caporaso et al., 2011). Because the same gene is sequenced for all targeted organisms in a community, the relatedness of different microbes can be inferred based on sequence similarity. As such, the first step in the analysis of targeted metagenomics is clustering similar sequences into groups which are called Operational Taxonomic Units (OTUs). An OTU can essentially be described as a "species cluster", where all sequences present in an OTU are similar and presumed to represent the same or very closely related organisms. The threshold for choosing what is or is not considered the same organism is dependent on the region that is being targeted for sequencing. For example, bacteria are clustered together at 97% identity, as the v4 16S rRNA region that is targeted is known to have ~3% intra-species variation. Once the sequences have been clustered together, each OTU is assigned a taxonomy based on similarity with sequences from known organisms (using databases specific for each targeted region). The taxonomy classification is very dependent on the presence of that organism in the databases that are being used. For example, if a novel bacterium is sequenced, no taxonomy is assigned because no representative would be present in the database. Furthermore, different levels of taxonomy can be assigned based on how confident or similar the sequence is to known organisms. An OTU can therefore sometimes be classified as "Bacteria", or a second OTU as "Bacteria;Proteobacteria" as the software pipeline is not confident in the assignment of lower taxonomy levels, while a third OTU may be classified as "Bacteria;Proteobacteria;Epsilonproteobacteria;Campylobacterales; Helicobacteraceae;Sulfurimonas" based on high similarity with other *Sulfurimonas* sequences found in the database. While we cannot always confidently assign names to each of these OTUs, we can be confident that they are different from one another, as we have already compared their sequences. Hence, even though in this example both the second and third OTUs belong to the *Proteobacteria* phylum, we are confident that they are in fact different organisms. Once taxonomy has been assigned, the abundance

of each OTU in each sample is determined. To compare the community profile of different samples, a multitude of statistical analyses were performed.

5. Results and Discussion

5.1. Site Assessment

The sections below correspond to results and discussion specific to each of the objectives that are reiterated here for ease of reference.

- Visual inspection along the toe of SWD, Main Dump, DSTSF, and area down-gradient of W37 to identify and document primary seeps that have the potential to be intercepted along with an assessment of current fate of seepage surface flows.
- Selection of a significant seep(s) for further test work based on initial site reconnaissance. Evaluate potential path of seepage interception trench in comparison to seep location(s).

Fourteen sites at the area receiving water from the SWD were selected for further testing (Table 1, Figure 2) and assessed for the parameters listed in Table 2. These include 3 seeps that are part of Minto's routine monitoring program (W32, W38, W39) as well as areas of the creek parallel to the seep (W38 Creek, W39 Creek, W32 Creek). Additionally, the W32 area had visible seepage between the seep and creek, and there were two distinctly different branches of the creek itself. Therefore, a total of four samples were collected at the W32 location. These are the W32 Seep, W32 Creek, W32 Mid (further seep between the toe seep and creek), and W32 Creek Red. The W32 Creek Red sample is a distinct branch of the W32 creek that was covered in red foam in August, and red precipitate on the soil in September (Figure 3). It should be noted that due to the parallel selection of sites and the direction of flow in the area between the seeps and creek, the water from seep W38 would never be present in W38 creek. We realize it is not possible to know the exact areas of entry of the seeps into the creek unless detailed hydrological studies and use of tracers were used; however, the purpose of this site assessment was to determine overall trends in water chemistry through the site and not a thorough hydrological investigation. During the first site assessment to locate priority sampling areas, the weather was hot (>+30°C daytime temperature) and dry, and therefore the seeps marked for routine monitoring (including W32, W38 and W39) were mostly dry with no visible standing or flowing water. However, these sites are routinely monitored and have a wealth of historical data relating to toe seepage from the SWD, and so they were identified as high priority areas. During the second site visit, snowmelt was causing a greater volume of seepage than in

the first visit. Water pools were present at all marked monitoring seepage points along the SWD toe, some with visible flows.

Runoff was observed directly from the road above via a culvert entering into the W15 collection pond, bypassing the creek and natural wetland area entirely (Figure 2). This "culvert runoff into W15" was sometimes clear, but at other times muddy, seemingly when the grader passed over the road on its routine schedule. While flow volumes and periodicity would need to be determined in order to calculate actual contribution to loading and concentrations, it is likely that water from the culvert (runoff into W15) is contributing to elevated concentrations of constituents of concern that have been observed to coincide with high total suspended solids (TSS) events at the W15 monitoring station at the far end of the pond (Figure 2).

Further geotechnical and hydrological assessments of the W15 area are planned to determine the depth of seepage, to ensure that the interception trench is adequately built such that seepage does not pass below it, and to delineate the extent of permafrost. This will be conducted under a separate scope of work.

The W37 area was inspected during the first site visit. However, as noted in the Phase V/VI project proposal, this area and the MVFE are likely to change with further development prior to closure. As such, they were not targeted as a key area for research at this time.

- Identify conceptual options for intercepting and routing seepage into the CWTS.

The W15 and W37 (Figure 1) areas are both in natural depressions and likely suitable for implementation of a CWTS, although this will also cause the depressions to receive non-contact runoff water from the general catchment area. Due to the multiple seeps along the toe of the SWD, an interception trench would be needed to collect the seepage and direct it towards the entry of the CWTS. Alternatively, it may be possible to construct several smaller CWTSs, each with their own entry point for seepage. These two options will be evaluated as information pertaining to flow rates and removal rates are refined through scientific testing and continued gathering of information from the site.

W37 area appeared to be amenable to future implementation of a CWTS. We recommend that studies be conducted on the sediments under the current water storage area behind the dam, in order to ensure there has been no accumulation of elements that might be released upon oxidation once the dam is breached upon closure.

- Identify probable locations for CWTSs. Investigation and evaluation of natural remediation processes that may be occurring on site as identified

by redox and pH changes between surface water and plant roots, and choice of plant types to be tested in Contango's pilot CWTS tests.

The W15, W37, and Main Pit have been identified as probable locations for CWTS in closure (Figure 1). The natural remediation processes occurring at the W15 site were investigated and evaluated in the second site assessment, with results and findings detailed in this report. We were unable to measure soil redox, however, as the water temperature was too cold for the reference probe, causing its internal buffer solution to crystalize (+0.5°C).

- Identification of borrow sites for CWTS substrate (limestone, coarse sand/gravel, peat, plant sources for much, etc.)

Overburden piles of sand and peat were located and tested for parameters relevant to their use as a soil in the CWTS. The samples of borrow material are suitable for use in a future CWTS. A source of slow release carbon (e.g., wood chips) is still yet to be identified.

- Identify ideal CWTS plant species based on site observations of vegetation.

Plant species and associated findings are discussed in section 5.3 below.

5.2. Water and Sediment Geochemistry

Dissolved chloride concentrations in the water can be used as a proxy to determine if concentration decreases in other elements can be attributed to dilution or putative treatment mechanisms. Comparison of changes in concentrations of elements of interest along the creek indicates fluctuations are similar to those of chloride, indicating dilution rather than treatment is occurring within the free flowing areas of the creek itself (Table 3). However, comparison of the water exiting and entering the W32 Creek Red site shows substantial treatment of copper and also some treatment of selenium and sulphur (Table 4). Comparison of the seeps with the creek also indicates significant treatment of copper, selenium and sulphur (Table 5). Overall, this indicates that active treatment of elements is occurring in the vegetated areas between the seep and the creek, but little or no treatment occurs in the creek itself. In this case, averages were used for the creek as the actual flow of seeps, and entry points into the creek are not known. More refined calculations cannot be made without detailed hydrological information for the area. Figures 4 and 5 show only small changes in chloride concentration through the system, yet large fluctuations in concentrations of other elements occur, especially near the seeps. Moreover, the chemistry of each seep varies greatly from one another, while the concentrations of elements in

the creek show more consistency (Figures 5-7), supporting the notion of treatment between the seep and creek, which evens out these fluctuations in chemistry by the time the water reaches the creek.

At the furthest upstream location sampled in this assessment, the W10 site has some of the highest copper concentrations in sediment (Figure 8), yet some of the lowest concentrations of copper in the water (Figure 6). This could reflect that the W10 site has been accumulating copper for a longer time span (i.e., in existence for more years, as it is a background location, upstream of the SWD), or perhaps the materials used to build the access road near the W10 contained copper and spread further than is currently visible due to overgrowth. This could also be caused by removal of copper from the water and into the sediment by binding to oxidized iron as suggested by the ratios of total vs. dissolved iron at that sampling area, with further stabilization (retention) of the copper promoted by conditions (redox, pH) in the sediment created by the combination of plants, microbes, and chemistry in the area.

The culvert runoff into W15 area has the highest copper concentrations in both sediment and water (Figures 6 and 8). The W15 runoff area has a separate and independent flow than the creek, entering the W15 collection pond via a culvert from the road above, bypassing the creek and natural wetland entirely (Figure 2). The culvert runoff into W15 appears to only flow during melt and rain events. This W15 runoff was sometimes clear (as in the time the sample for this study was taken), but at other times is muddy, seemingly when the grader passed over the road on its routine schedule. While flow volumes and periodicity would need to be determined in order to calculate actual contribution to loading and concentrations, it is likely that water from the culvert (W15 runoff) is contributing to elevated concentrations of constituents of concern that have been observed to coincide with high TSS events at the W15 monitoring station at the far end of the pond. This can be seen in Figure 9 as the W15 inflow (from creek) has a lower concentration of key elements (e.g., copper, selenium) than is routinely measured at the W15 monitoring point. The only visible source for this loading is the culvert runoff into W15. This is a key finding of this site assessment, as the routine sampling performed at the set W10 and W15 monitoring points would indicate that there is less treatment ongoing in the natural wetland area than is truly occurring. This is because of the extra loading of contaminants contributed by the culvert runoff into W15 mixing in the collection pond with the cleaner water exiting the natural wetland area before reaching the W15 monitoring point (Figures 6 and 7).

The W32 Creek Red site is a location between the W10 (upstream) and W15 (downstream) areas where the creek briefly splits into two steams at the W32 clearing. The W32 Creek Red sample has a high concentration of oxidized iron, indicated by a visible red precipitate (Figure 3). At the first site visit in August, the red substance was fluffy and floating in the water and

also on the sediment. In September, the red substance was entirely at the sediment interface. Moreover, the water exiting the W32 Creek Red site has the lowest concentration of copper in the water (at 1.92 µg/L, the only site sampled in this study with copper concentration below CCME guidelines). This finding, paired with moderate concentrations of copper in the sediment at the W32 Creek Red site compared to other locations (Figure 8), suggests that copper removal may be occurring via binding to oxidized iron at the W32 Creek Red site. It would be useful in the future to monitor sediment redox at this and other sites; however, the water temperature (+0.5°C) was too cold for the reference probes to be used at the time of the September sampling event. There were several other smaller areas similar in appearance to the W32 Creek Red site found along the creek, always where water was shallow and *Equisetum* or *Carex* present.

The W32 Seep has low ammonia (9.4 µg/L) and high nitrate (39.9 mg/L) concentrations in the water relative to other samples (Figure 7). It appears that the ammonia is being oxidized to nitrate as it seeps through the waste rock pile, which is consistent with the peaks of sulphur and selenium concentrations in the water at the W32 site, both of which could be released from the waste rock under oxidizing conditions. In contrast, the W32 Creek and W32 Creek Red water samples have higher concentrations of ammonia and lower concentrations of nitrate than other samples (measured as N, 0.11 and 0.18 mg/L ammonia and 3.02 mg/L and below detection limit (<0.02 mg/L) for nitrate, respectively; Figure 7). It is possible that there are greater concentrations of ammonia seeping from the SWD between the W10 site and W32 creek sampling point, or that the ammonia is being produced within the creek from decaying organic material. It is also possible that the ammonia is being produced by the abundant *Geobacter* bacteria at the W32 Creek Red site (see section 5.4), as it is a known iron-reducing organism that feeds on electrons from oxidized iron and can also respire nitrate to ammonia (Lovley & Phillips, 1988; Senko & Stolz, 2001).

As would be supported by redox and biogeochemical theory, there is a strong correlation between ammonia and total iron concentrations in water (Figure 10), while iron and nitrate show a negative correlation. This makes sense from a theoretical oxidation-reduction chemistry perspective, as ammonia will oxidize into nitrate, increasing nitrate concentration while the same oxidizing environment will cause iron to oxidize and precipitate from solution. Conversely, the sulphur, selenium, and copper concentrations in water are positively correlated to the nitrate concentration in water (Figure 10), as would be expected since sulphur and selenium are normally present as soluble compounds in oxidizing environments. Likewise, at increasingly

negative redox nitrate will be reduced to nitrogen gas, followed by selenate/selenite to elemental selenium and at an even lower redox, sulphate is reduced.

5.3. Plants

Sphagnum (Moss) was found to grow in nearly all areas of the site, except directly at the outlet of the creek to the W15 pond. *Salix sp.* (Willow) was found at all sites, except at sites in the W32 area, which appeared to have been cleared at some point. *Carex aquatilis* (Water Sedge) was dominant within the W32 area that was recently cleared (Figure 11), and also within the W10 area where the water was deeper. *C. aquatilis* was found to be spreading through the W10 area access road, being the first plant to colonize the road and even recent tire ruts (Figure 12), indicating that it will be a potentially good choice for quickly establishing the CWTS. However, at locations with shallower water, *C. aquatilis* and *Sphagnum* coexisted, and *Equisetum* was also present, growing through the *Sphagnum.* This suggests that if water depth is not maintained, *Sphagnum* and *Equisetum* will grow in succession.

Compared to the other plant species tested, *Sphagnum* showed significantly greater uptake of elements such as copper, uranium, and iron at all sites that *Sphagnum* was available (Figures 13 and 14). No detectable selenium was found in any of the plants except for *C. aquatilis* and *Sphagnum* at the W32 Seep site (0.7 mg/kg) and *Sphagnum* at the W32 Mid site (0.3 mg/kg). This is presumably due to higher amounts of selenium being present in water at the W32 Seep site (Figure 6). Uptake of copper and iron in all plants was negatively correlated with copper concentration (-0.904) or iron concentration (-0.994) in sediment at each location, respectively. These findings suggest varying bioavailability of copper at the different sites sampled, and when elements are being deposited into the soils, they are in forms (species) that are less bioavailable to the plants than when in aqueous form.

All cases of water channeling through the natural wetland area were in association with *Salix* plants (Figure 15). *Salix* species had the least uptake of elements compared to other plant species at most of the sites (Figure 13). The root structure of these plants appears to contribute to the creation of channelized creeks, which also corresponds with little or no treatment occurring in the creek area. *Salix* has not regrown in the W32 area that was previously cleared, and it is recommended that this area be surveyed yearly

to assess whether they will begin to reestablish in this area. It should also be determined when the W32 area was cleared. This information will be used to establish a maintenance schedule for the wetlands, as young *Salix* plants should be removed before becoming well established.

Although a greater water depth is one parameter that can be used in CWTS to create an anaerobic redox and decrease dissolved oxygen concentration in the water, the research conducted here has shown that *Sphagnum* can be an effective means of removing elements from the seepage through uptake, a finding supported by literature. As such, this potential succession of plants may occur, but is not expected to significantly impact the performance of the CWTS, and may even improve the performance via plant uptake in the *Sphagnum*.

5.4. Microbiology

Genetic profiling of the microbial communities indicate great diversity with hundreds of species, but for the purpose of discussion in this section, the top ten most abundant organisms are focused upon (Appendix I). When assessing the microbial genetic profiles of all sites, most samples have a very similar distribution of organisms (Figure 16). The outflow of the W15 creek was sampled in triplicate as a control for microbial variability within a given sampled site. All of the creek samples receiving seepage (W32, W38, W39, and the three W15 sites) cluster together, which indicates similar microbial profiles. All three W15 sites show nearly identical microbial distributions (Appendix I and II), while strong similarities are found for all creek sites. One common trend in all creek sites is the presence of the *Intrasporangiaceae* organism group (yellow in Appendix I). This group appears to be closely related to *Tetrasphaera,* which are known to be involved in phosphorus removal from wastewater. Conversely, the W10 and W10 *Carex* Root samples show distributions more similar to one another than to the downstream creek locations. A large distribution of the organisms at W10 and W10 *Carex* Root locations represents the order *Bacteroidales*, which are ubiquitous colonizers of a wide range of environments (e.g., soil, activated sludge, decaying plant material, compost, fresh water and marine samples, dairy products). However, the W10 *Carex* Root had a denser population of organisms than the W10 sample taken away from plants in the area (i.e., # of organisms per gram of sample; Figure 17).

Laboratory growth-based assays also tested for total heterotrophic, selenite-, sulphate-, and nitrate-reducing organisms at different sites (Figures 17 and 18). The population of total heterotrophs grown under aerobic conditions was

significantly greater than under anaerobic conditions (Figure 18). Interestingly, both the laboratory- and genetic profiling-based methods showed very few sulphate-reducing organisms at most of the sites, with the highest amounts found at the W10 *Carex* Root and W15 inflow sites (Figure 17). The W10 *Carex* Root sampling location also showed interesting sediment features with black color and smell at deeper depths, characteristic of sulphate-reducing organisms, but with red iron oxides visible at the water interface (Figure 19). Selenite-reducing organisms were found at all tested locations, while nitrate-reduction only occurred to completion at the W10 *Carex* Root, W10, W32 Creek Red, W38 Creek and W15 inflow locations (Figure 17). No correlation was found between the number of selenite-, sulphate-, or nitrate-reducing organisms and the water concentration of nitrate, selenium, and sulphate at each location. This is potentially due to the correlation analysis being performed on water concentrations instead of sediment concentrations for these compounds (which is where the reducing organisms would be found), as non-detectable selenium and nitrogen levels were found in most of the sediment samples. In the case of sulphur, the laboratory (Maxxam Analytics) did not perform the requested analysis on the soils, and discussions are ongoing to determine if new samples could be collected and analyzed at their cost to fill in this data gap.

In addition to the red precipitate at the W32 Creek Red site, this location has a unique microbiological profile (Figure 16 and Appendix I). A very dominant percentage of the microbial population at W32 Creek Red is a bacterium called *Geobacter*. This organism is known to seek out insoluble iron oxides, and establish direct contact and reduce them as part of its metabolism. As such, it is likely that the high abundance of *Geobacter* bacteria found at W32 Creek Red is actively reducing Fe(III) to Fe(II), which then reduces Cu(II) to Cu(I) through a coupled biogeochemical reaction. In addition to the interplay between iron-reduction and other elements, the W32 Creek Red site has much lower available phosphorus in the soil and total phosphorus in the water in comparison to the W32 Creek site which is water flowing just adjacent to the red precipitate area, also suggesting that increased microbial activity may be playing a role.

6. Summary and Conclusions

Collectively, these findings contribute greatly to the ability to design and operate a CWTS for effective treatment of waters at the site. The geochemical data indicate that substantial treatment of seepage is occurring on site within the natural wetland, validating the concept that a constructed wetland could be designed to function in the long-term closure condition at

the site, but with optimized treatment performance. Mechanisms for treatment of copper and selenium were identified in the natural wetlands receiving seepage from the SWD, including dissimilatory sulfate reduction followed by sulfide precipitation, dissimilatory selenium reduction, and iron co-precipitation.

It is clear that there is greater and faster treatment occurring within the natural wetland than would have been evident if calculated only using data collected by previously existing routine sampling programs. The additional loading of elements at the runoff/culvert into the W15 collection pond is likely a significant contributor to loading during rainfall and melt events, masking the treatment that is occurring through the wetland for seepage that is originating from the SWD toe.

It is evident that the seepage chemistry varies greatly, both between seeps and over time (this study and historical data gathered). This will need to be taken into consideration when evaluating options of constructing a seepage diversion trench compared with building multiple smaller wetlands. The sources and amounts (frequency and range in flows) can be coalesced or treated as separate flows with individual specifically designed wetlands. Collection of seepage by a diversion trench would allow mixing of the various seepage chemistries and more uniform composition and volume of inflow water to the wetlands, but individual wetlands may be favoured in the W15 area due to topography and seepage locations.

There appears to be sufficient soil on site for both hydrosoil and berms for the CWTS. Analysis indicated that the chemistry of overburden soils is amenable to use as soil in the CWTS and to support plant growth, although slightly elevated in copper as is the entire background area. The overburden peat on site is also amenable to use in the soil as an amendment to the CWTS. However, additional sources of organic matter to create anaerobic conditions in the sediment and improve dissimilatory reduction pathways for sulfur and selenium will still need to be identified (such as wood chips, straw, and/or compost). Additionally, when an onsite demonstration-scale CWTS is built, there should be a mechanism developed to deliver seepage water to the CWTS rather than creek water, as substantial treatment is currently occurring between the seeps and the creek.

There is suitable vegetation present that is well adapted to the site conditions. The plants at site show a robust ability to grow in the range of water chemistries present in the area. Moreover, all sites sampled are host to a wide diversity of microbes with known capability of performing

favourable biogeochemical reactions, which will contribute to water treatment in a CWTS. This confirms that even at sites where no chemical evidence of treatment was found, there are microbes present that would be capable of these reactions if encouraged with the right environment (e.g., plant roots, shift in dissolved oxygen or redox, availability of electron donors). While organisms capable of favorable geochemical reactions were ubiquitous at the site, they were found in elevated numbers associated with *Carex* roots, enforcing our selection of this plant species for the CWTS.

CWTS designs are currently being piloted at Contango's year-round pilot facilities. These systems are planted with *C. aquatilis* from site, and peat from the selected borrow site is incorporated into the soils. Replicate pilot-scale CWTS experiments are in progress with and without *Sphagnum* added in order to assess whether the *Sphagnum* contributes positively or negatively to the treatment, and also whether succession from *C. aquatilis* to *Sphagnum* as the dominant plant can be prevented or enhanced through modification of water depth.

The pilot study will continue to integrate information and design decisions, as well as provide supplementary soil and water redox measurements to validate hypotheses about biogeochemical pathways in the CWTS. Further to this, pilot CWTS sediments will be subjected to leachabilty testing for elements of concern to confirm suitability and longevity of the designs. CWTS designs will continue to be optimized as closure water chemistry and volumes are refined.

7. Closure

We trust the information herein satisfies your present requirements. Should you have any questions, please contact the persons listed below. We appreciate the opportunity to provide the services detailed in this report, and look forward to discussing any comments you may have.

Respectfully submitted,

Contango Strategies Ltd

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9. Tables

Table 1: Summary of samples taken at the Minto Mine site

Location	Water Sample	Sediment Sample	Micro Sample	Plant Sample
W10 *	Yes	Yes	Yes	Yes
W10 *Carex* Root	-	-	Yes	-
W32 Seep	Yes	-	-	Yes
W32 Mid	Yes	-	-	Yes
W32 Creek	Yes	Yes	Yes	Yes
W32 Creek Red	Yes	Yes	Yes	-
W39 Seep	Yes	-	-	-
W39 Creek	Yes	Yes	Yes	-
W38 Seep	Yes	-	-	-
W38 Creek	Yes	Yes	Yes	-
W15 Inflow 1	Yes	Yes	Yes	Yes
W15 Inflow 2	-	-	Yes	-
W15 Inflow 3	-	-	Yes	-
Culvert runoff into W15	Yes	Yes	Yes	-

* *Carex aquatilis* and *Sphagnum* for the Minto Pilot 1 CWTS at Contango Strategies were obtained from this location

Table 2: Summary of water, soil, and plant analyses

Water Analysis	**Soil Analysis**	**Plant Analysis**
Total Regulated Metals (Al, Sb, As, Ba, Be, Bi, B, Cd, Cr, Co, Cu, Fe, Pb, Li, Mn, Mo, Ni, P, Se, Si, Ag, Sr, Tl, Sn, Ti, U, V, Zn, Zr, Ca, Mg, K, Na, S, Hg)	Assessment ICP Metals (Al, Sb, As, Ba, Be, Bi, Cd, Ca, Cr, Co, Cu, Fe, Pb, Li, Mg, Mn, Hg, Mo, Ni, P, K, Se, S, Ag, Na, Sr, Tl, Sn, Ti, U, V, Zn, Zr)	Metals (Al, Sb, As, Ba, Be, Bi, B, Cd, Ca, Cr, Co, Cu, Fe, Pb, Mg, Mn, Mo, Ni, P, K, Se, Ag, Na, Sr, Tl, Sn, Ti, U, V, Zn)
Routine Water (Alkalinity, HCO_3, Ion Balance, Dissolved ions, Hardness)	Available Nitrogen, Phosphorous, and Potassium (NPK)	
Total Suspended Solids (TSS)	Cation Exchange Capacity (CEC)	
Chemical Oxygen Demand (COD)	Sodium Adsorption Ratio (SAR)	
Biological Oxygen Demand (BOD)	Total Organic Carbon (TOC)	
Total Kjeldahl Nitrogen (TKN)	Electrical Conductivity (EC)	
Total Organic Carbon (TOC)	Texture Analysis (Physical properties)	
Ammonia		

Table 3: Copper, selenium, sulphur, and chloride concentrations along the creek. The approximate fold change in concentration along the creek is provided in grey in parentheses.

	W10	**W32 Creek**	**W38 Creek**	**W39 Creek**	**W15 Inflow**
Total Copper (µg/L)	5.92	32.5 (5.5)	10.6 (0.3)	11.5 (1.1)	19.7 (1.7)
Total Selenium (µg/L)	<0.1	0.71 (7.1)	2.64 (3.7)	2.53 (1.0)	2.10 (0.8)
Total Sulphur (mg/L)	<3	9.7 (3.2)	24.7 (2.5)	23.0 (0.9)	23.1 (1.0)
Dissolved Chloride (mg/L)	1.2	1.8 (1.5)	3.0 (1.7)	3.1 (1.0)	2.9 (0.9)

Table 4: Copper, selenium, sulphur, and chloride concentrations at W32 Creek compared to W32 Creek Red location, and W32 Mid compared to W32 Seep. Approximate fold concentration difference between the two sites is provided in grey parentheses.

	W32 Creek	**W32 Creek Red**	**W32 Seep**	**W32 Mid**
Total Copper (µg/L)	32.5 (0.06)	1.93	10.1 (1.5)	15.1
Total Selenium (µg/L)	0.71 (0.3)	0.21	19.6 (0.01)	0.24
Total Sulphur (mg/L)	9.7 (0.3)	<3	110 (0.1)	11.9
Dissolved Chloride (mg/L)	1.8 (2)	3.5	5.5 (1.2)	6.8

Table 5: Average copper, selenium, sulphur, and chloride concentrations in the creek (W32, W38, W39, and W15; i.e., not including W10), compared to each seep. Approximate fold concentration difference between the seep and average creek is provided in grey parentheses.

	Avg. Creek	**W32 Seep**	**W38 Seep**	**W39 Seep**
Total Copper (µg/L)	18.6	10.1 (0.5)	102 (5.5)	71.4 (3.8)
Total Selenium (µg/L)	2.0	19.6 (9.8)	7.59 (3.8)	0.94 (0.5)
Total Sulphur (mg/L)	20.1	110 (5.5)	45.8 (2.3)	46.0 (2.3)
Dissolved Chloride (mg/L)	2.7	5.5 (2.0)	5.5 (2.0)	3.8 (1.4)

10. Figures



Figure 1: Conceptual closure diagram of Minto mine (Appendix A, Minto Phase V/VI Preliminary Reclamation and Closure Adaptive Management Plan), indicating potential wetland locations and routine monitoring sites.



Figure 2: Sampling locations at the Minto mine. All routine monitoring points are indicated in white, while sampling locations unique to this study are shown in yellow. The blue arrows indicate the direction of water flow. The W32 Creek label corresponds to the location of both the W32 Creek and W32 Creek Red.



Figure 3: Minto Site W32 Creek (bottom in A; left in B) and W32 Creek Red (top in A; right in B) in A) August 2013 and B) September 2013. The pictures show the same site at two different dates.

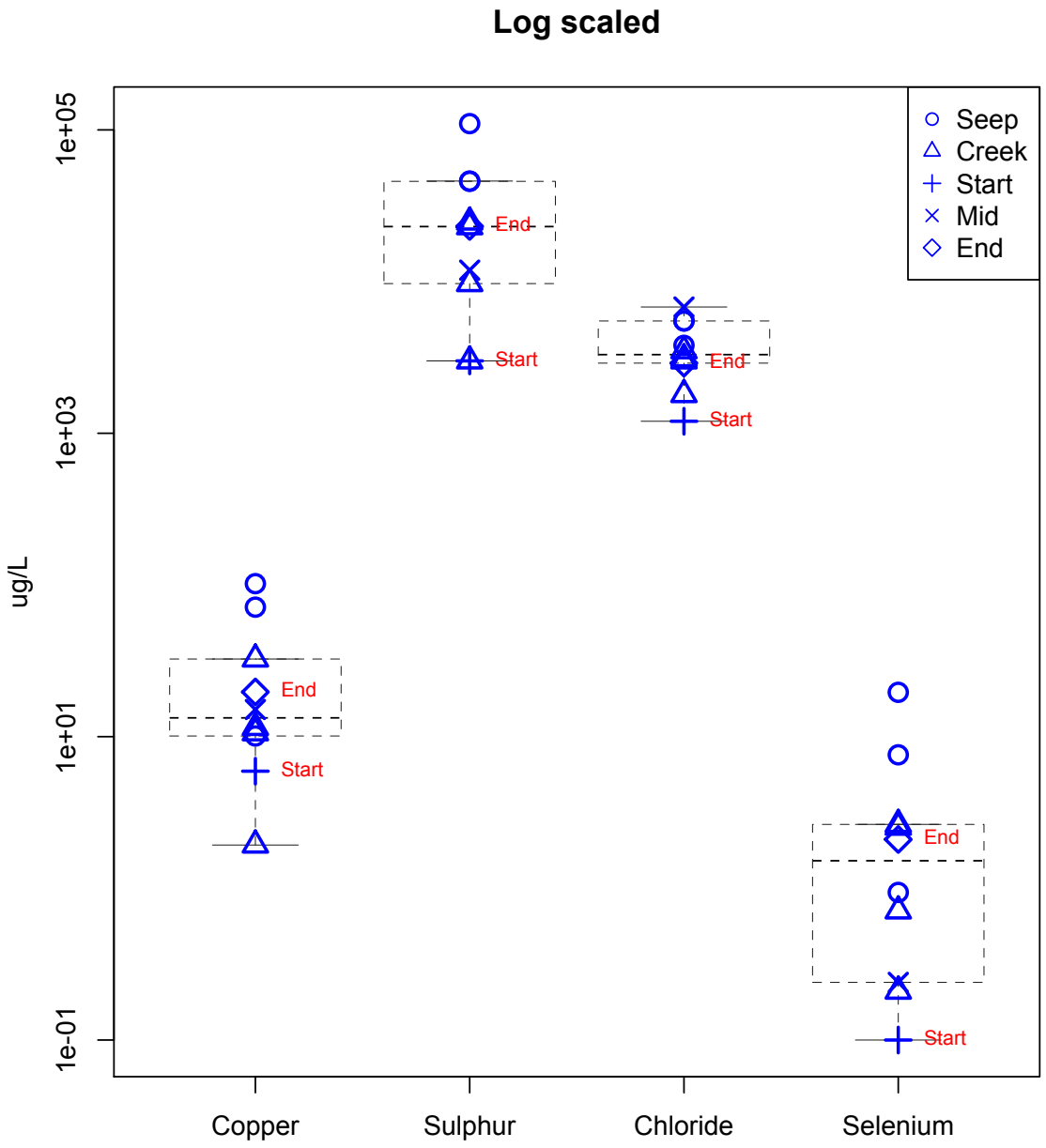


Figure 4: Comparison of copper, sulphur, chloride and selenium concentrations in the creek between the W10 and W15 areas, and seeps along the toe of the SWD. "Start" indicates the W10 location, while "End" indicates the W15 inflow. The W32, W38, and W39 Creek locations are shown as triangles, while all seeps are differentiated as circles. The W32 Mid site is depicted as a X. Note the spread of the copper, sulphur, and selenium plots is greater than that of chloride. The seeps demonstrate the highest concentration of copper, sulphur, and selenium, while chloride concentrations do not follow this trend.

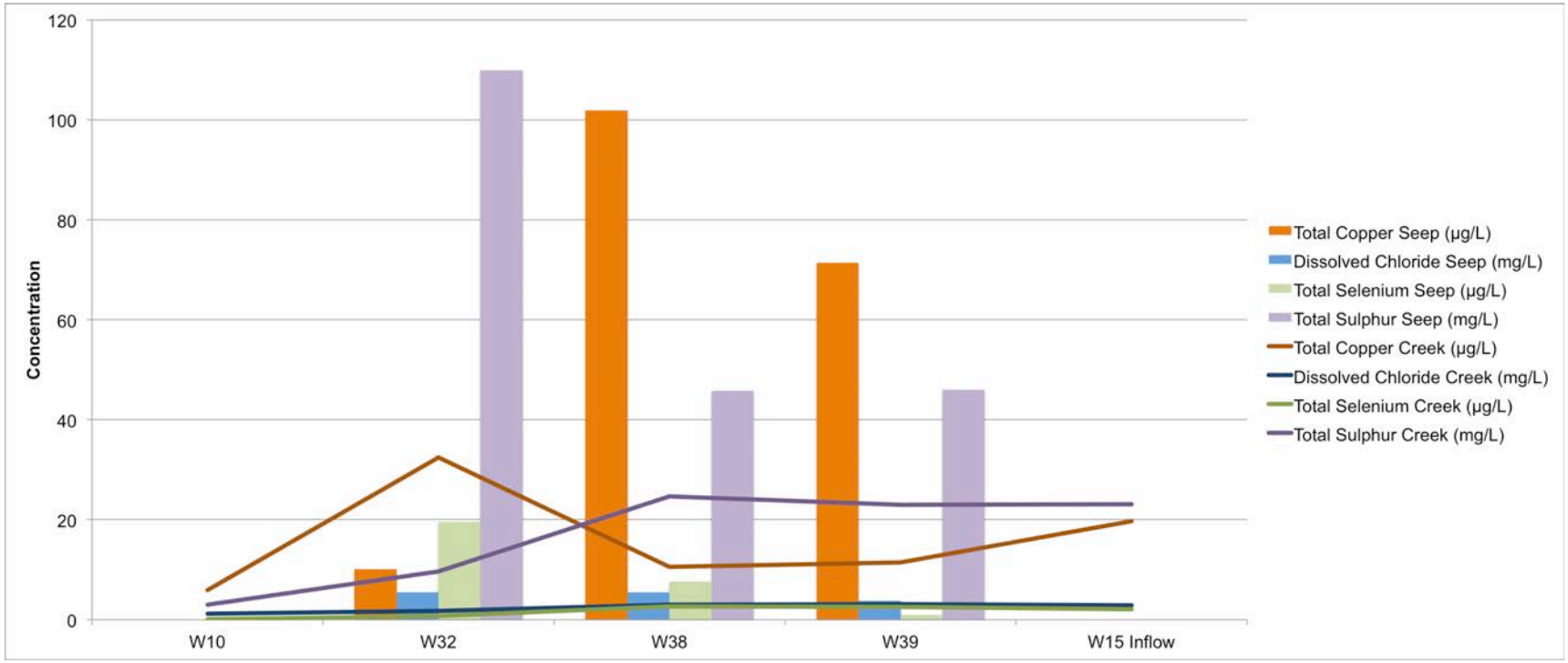


Figure 5: Copper, chloride, selenium, and sulphur concentrations in the creek and seep locations. Bars indicate seepage, while lines indicate the creek concentration of each element.

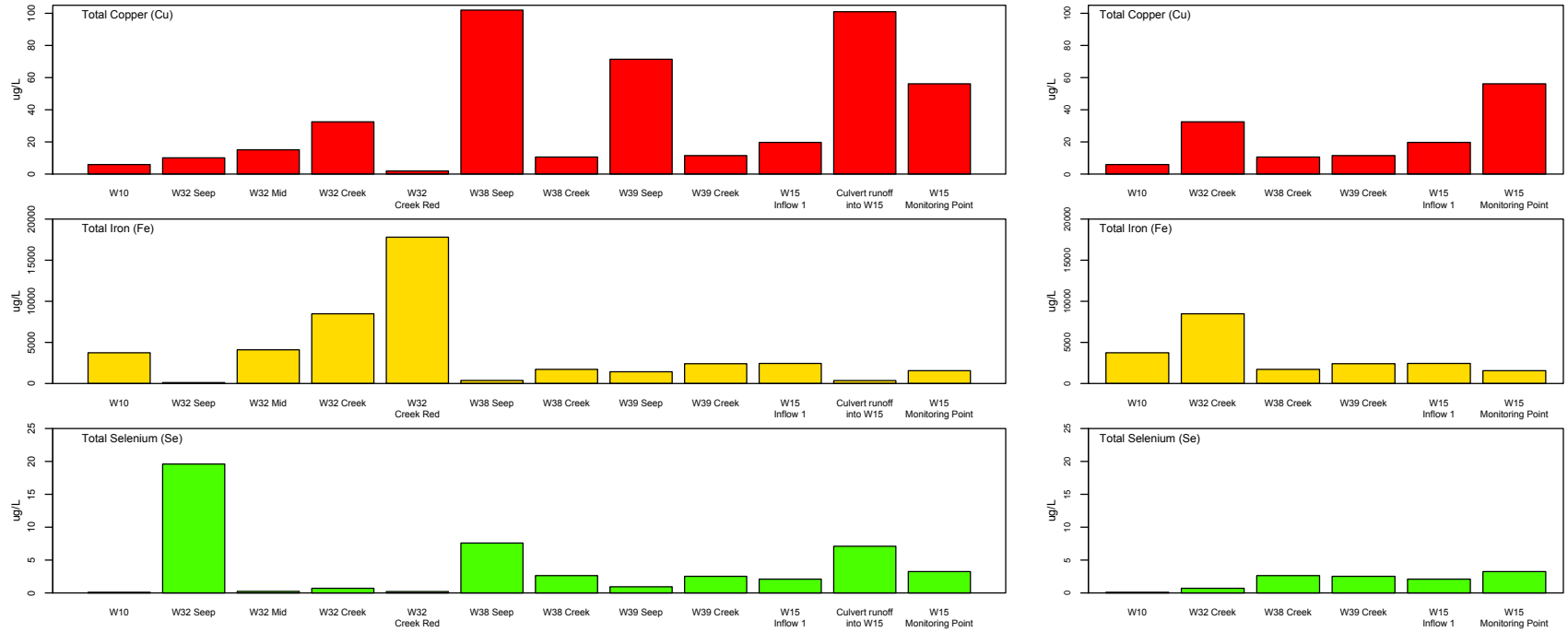


Figure 6: Copper, iron, and selenium concentration in water. The left plots show all sites, while the right plots show the distribution of elements along the creek. The W15 Monitoring Point represents the average of September 17th and September 28th, 2013, monitoring timepoints, which are the closest dates to the site visit when samples were obtained.

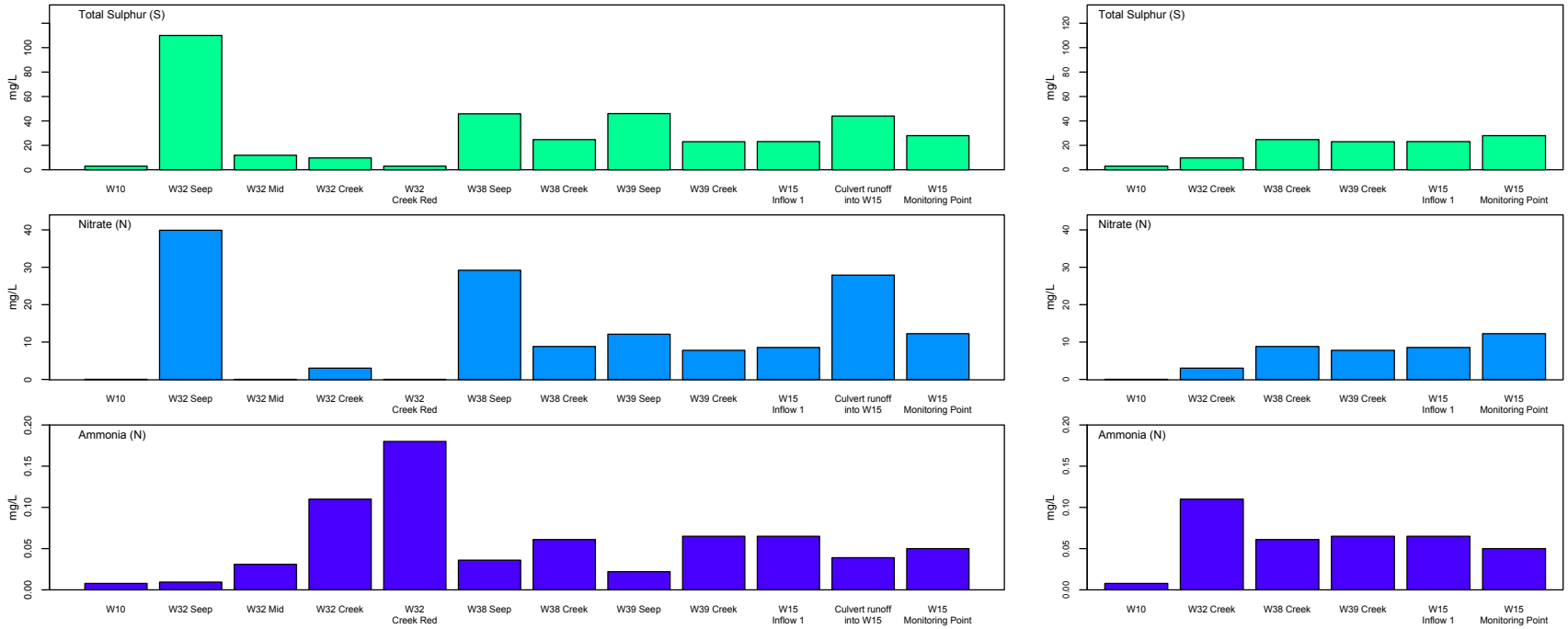


Figure 7: Sulphur, nitrate, and ammonia concentration in water. The left plots show all sites, while the right plots show the distribution of elements along the creek. The W15 Monitoring Point represents the average of September 17th and September 28th, 2013, monitoring timepoints, which are the closest dates to the site visit when samples were obtained.

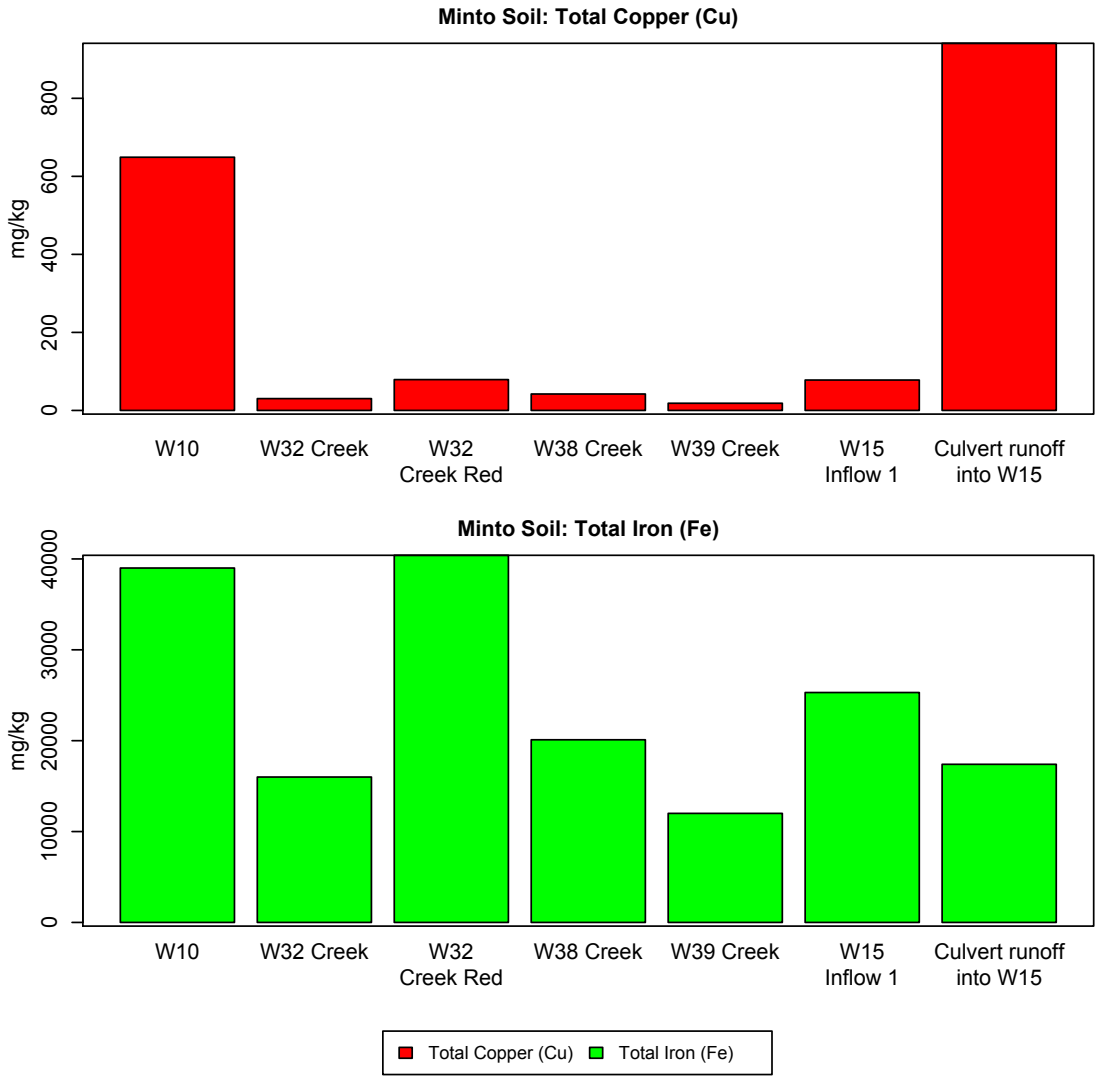


Figure 8: Copper and iron concentration in soil. Selenium was below the detection limit in sediment (<0.5 mg/kg), and is therefore not shown.

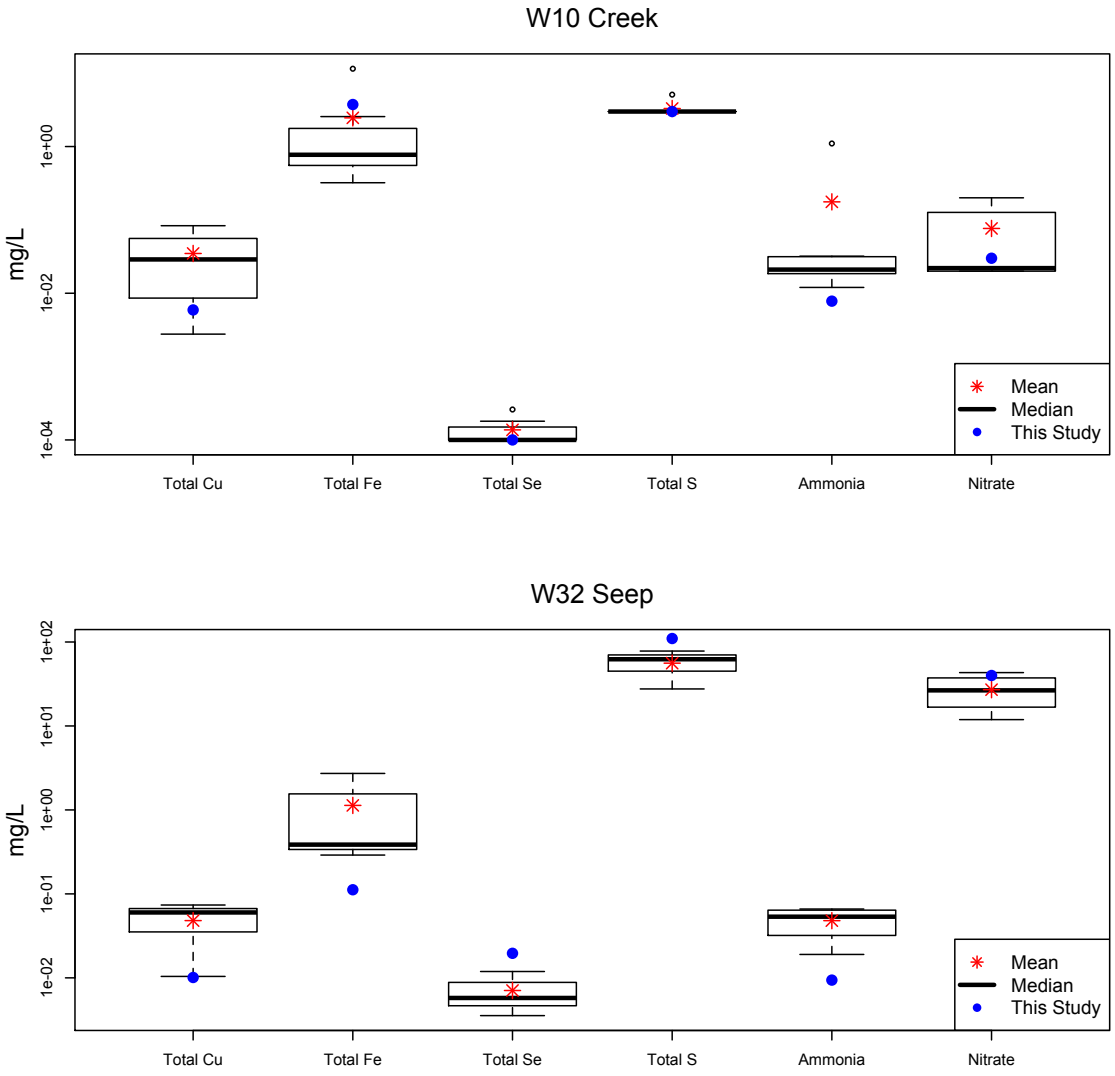


Figure 9: Concentration of select elements from all 2013 routine monitoring program measurements. Blue dot indicates concentration measured in samples collected as part of this study. Figure is continued on following page.

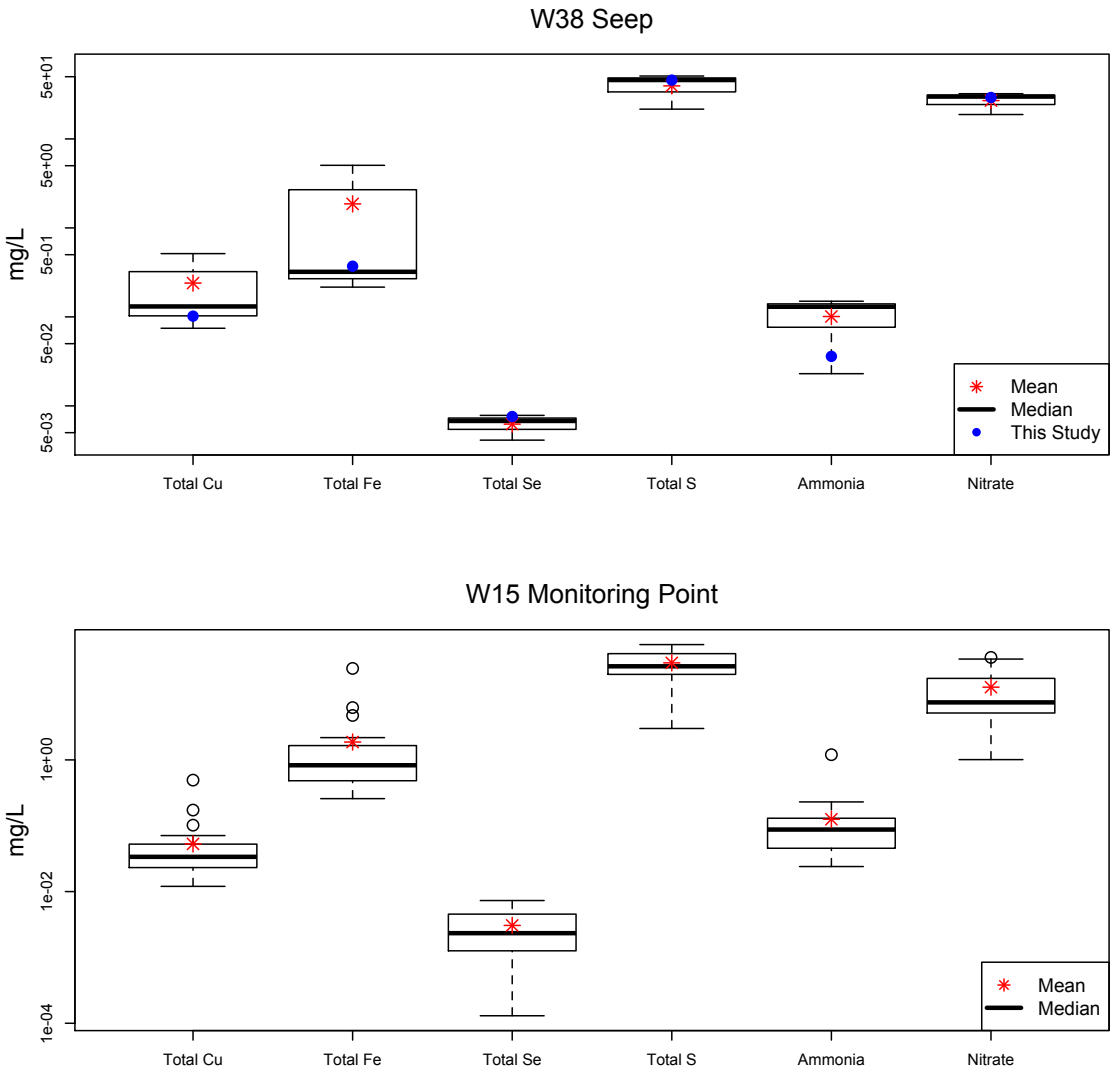


Figure 9: Concentration of select elements from all 2013 routine monitoring program measurements. Blue dot indicates concentration measured in samples collected as part of this study. Figure is continuation from previous page.

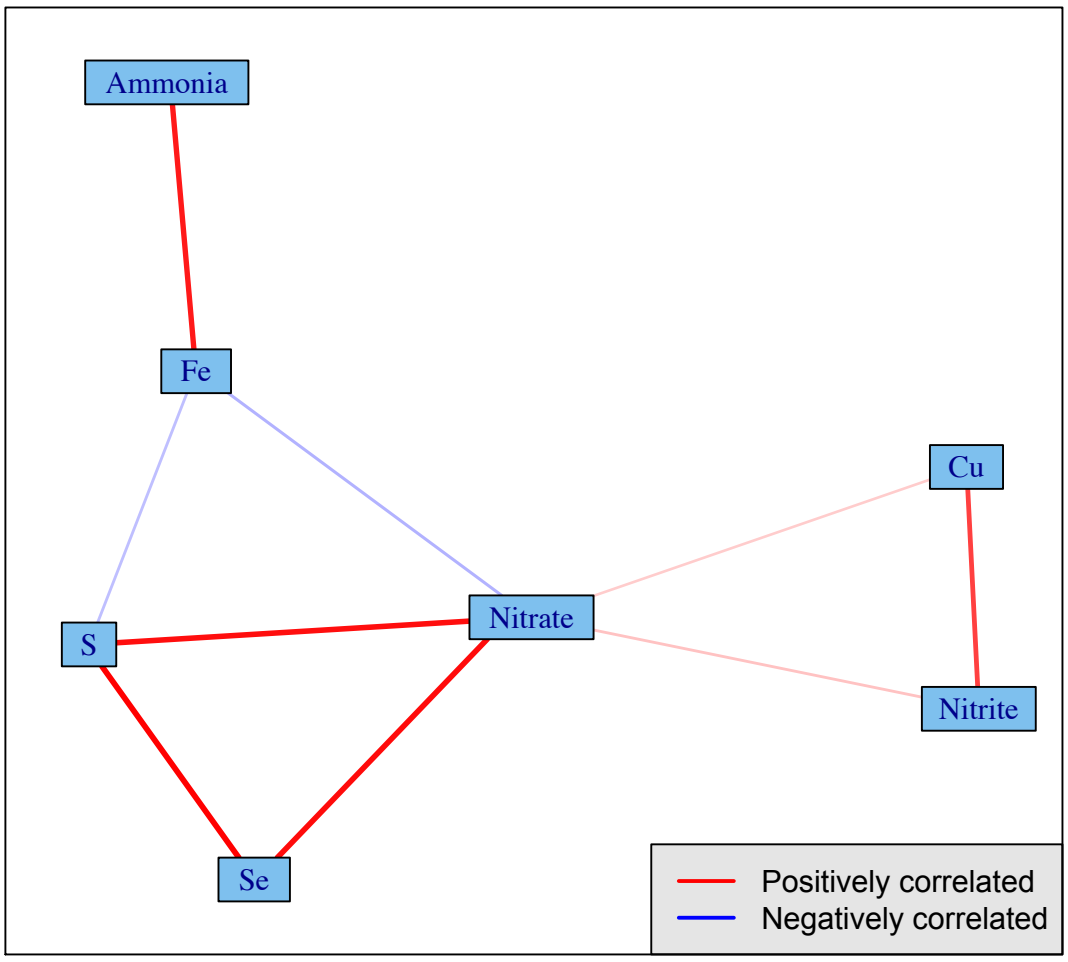


Figure 10: Correlation of various compounds in water. The strength of the line connecting different compounds indicates the strength of the correlation. For example, the thick line between Se and S corresponds to a Pearson correlation coefficient of 0.939, while the line connecting Cu, and Nitrate or Nitrite corresponds to a coefficient of 0.527 or 0.812, respectively. All Pearson correlations are significant at a level of P<0.1. Also of note, copper concentrations in soil were found to be significantly correlated with copper and nitrite concentrations in the water (not depicted in figure).



Figure 11: *C. aquatilis* and grasses are dominant at the W32 area that was recently cleared.



Figure 12: *C. aquatilis* spreading through the W10 area access road, being the first plant to colonize the road and recent tire ruts.

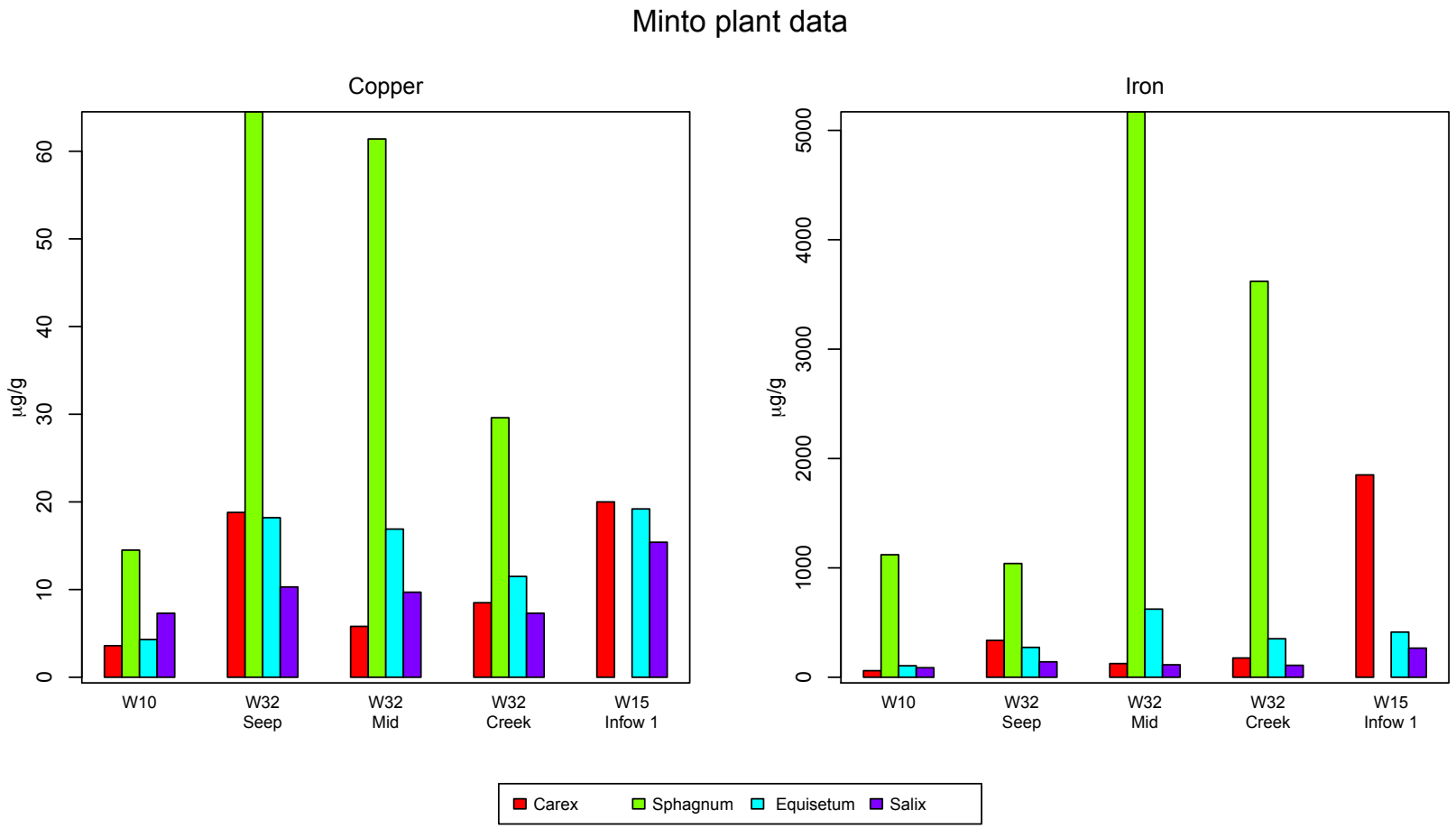


Figure 13: Concentration (µg/g) of dry plant matter for copper and iron in plants at the Minto Site. Note that there was no *Sphagnum* present at the W15 inflow site available for collection.

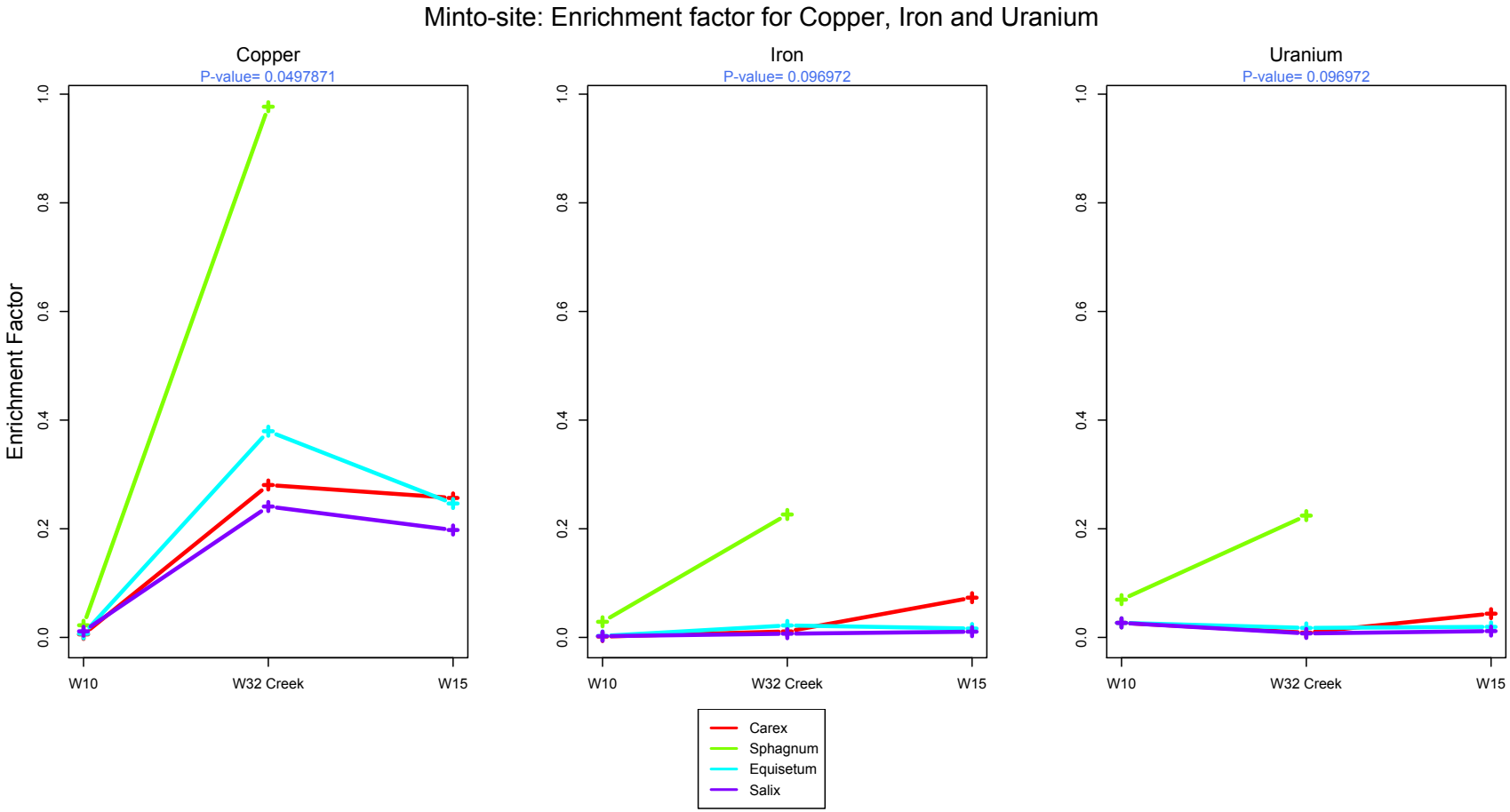


Figure 14: Enrichment factor for copper, iron, and uranium in *Carex, Sphagnum, Equisetum,* and *Salix.* A p-value <0.1 indicates there is a significant difference in the enrichment factor at different sites. Note that there was no *Sphagnum* present at the W15 inflow site available for collection.



Figure 15: Example of creek channeling through *Salix* plants.

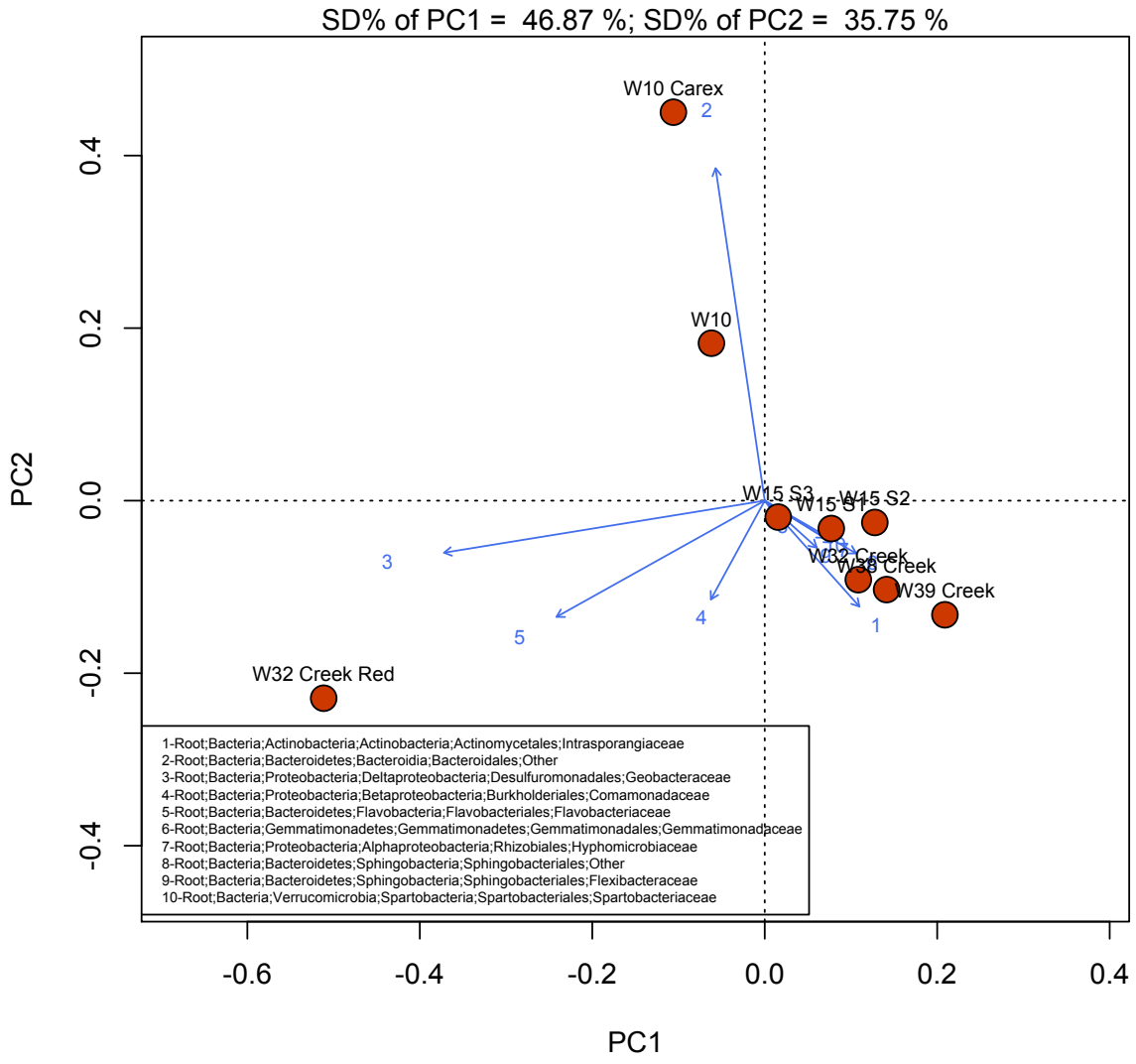


Figure 16: Principal component analysis of all sites based on microbial community profiles. All sites that cluster together have similar microbial profiles, while sites clustering on their own (e.g., W32 Creek Red) show a less similar profile. Each numbered arrow represents a group of microorganisms listed in the legend. Essentially, the direction and length of an arrow represents how abundant that group of microorganisms is in each sample. For example, W32 Creek Red would show a high abundance of *Geobacteraceae* (arrow #3) and *Flavobacteriaceae* (arrow #5), while W10 Carex shows a high abundance of *Bacteroidales* (arrow #2). All creek samples (W32, W38, W39, and W15) cluster together and show similar microbial profiles (high abundance of *Intrasporangiaceae,* arrow #1).

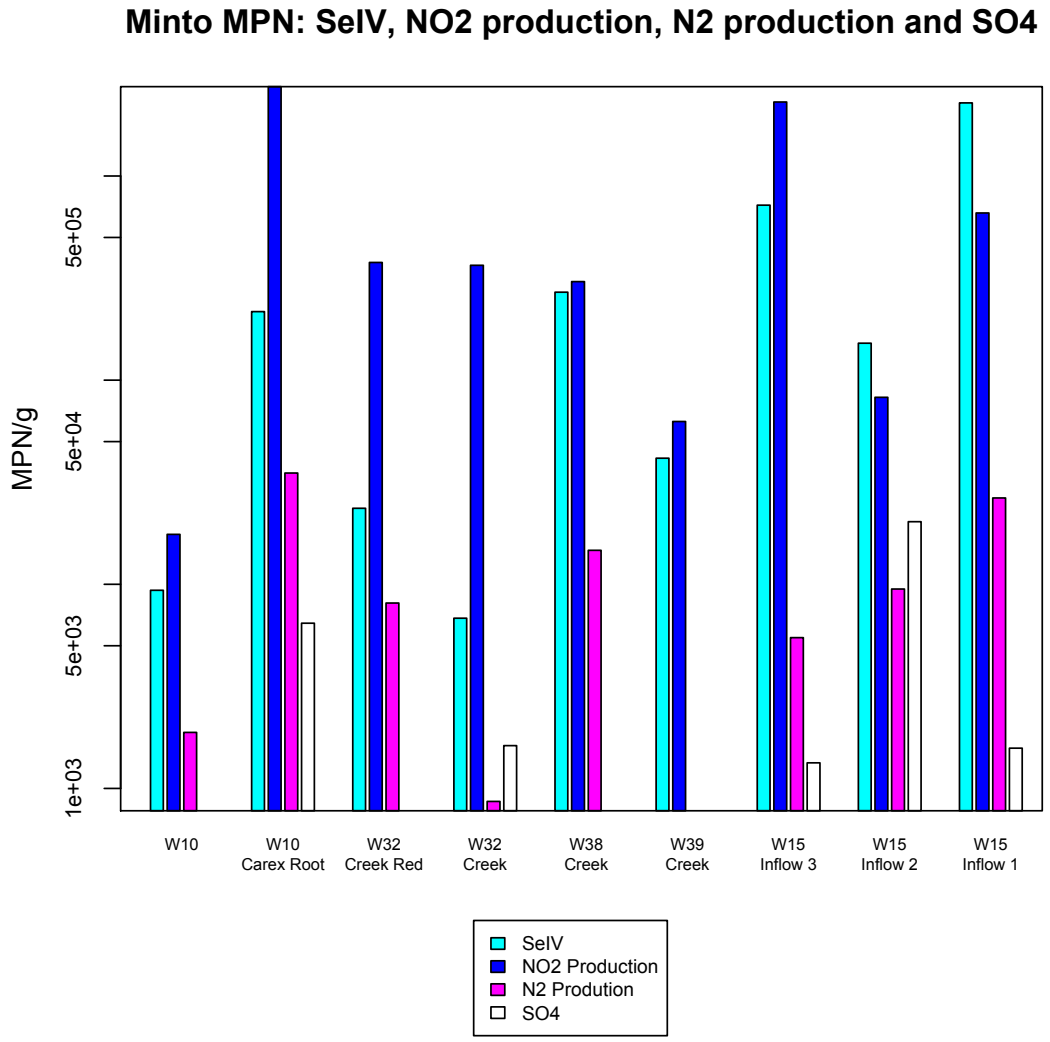


Figure 17: Predicted number of selenite, sulphate, and nitrate reducing bacteria found at various Minto sites, based on the most-probable number assay. The nitrate-reduction assay tests for microbes that can reduce nitrate to nitrite (i.e., NO_2 Production), as well as microbes that can further reduce nitrite to nitric oxide or nitrogen gas (i.e., N_2 Production).

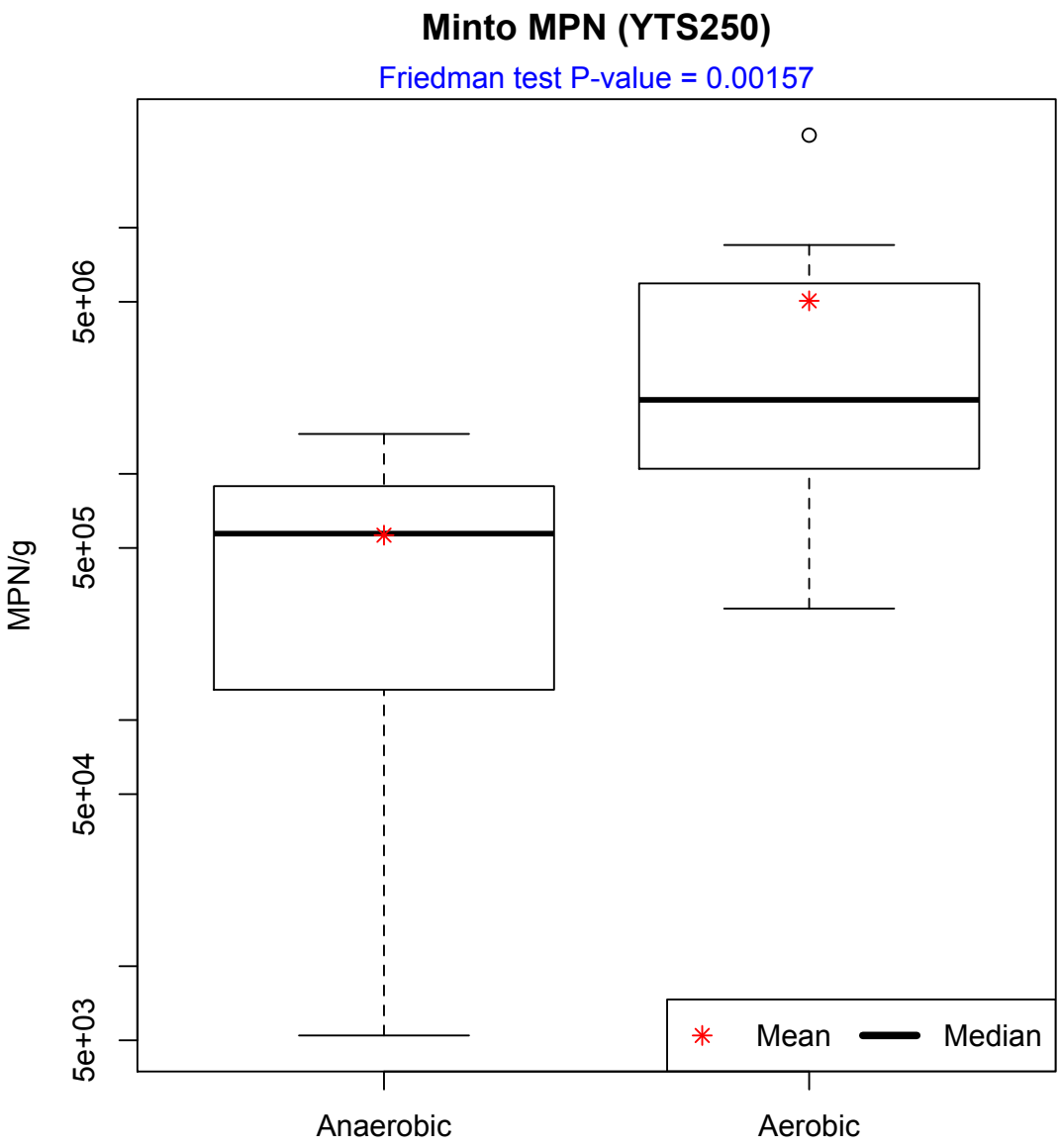
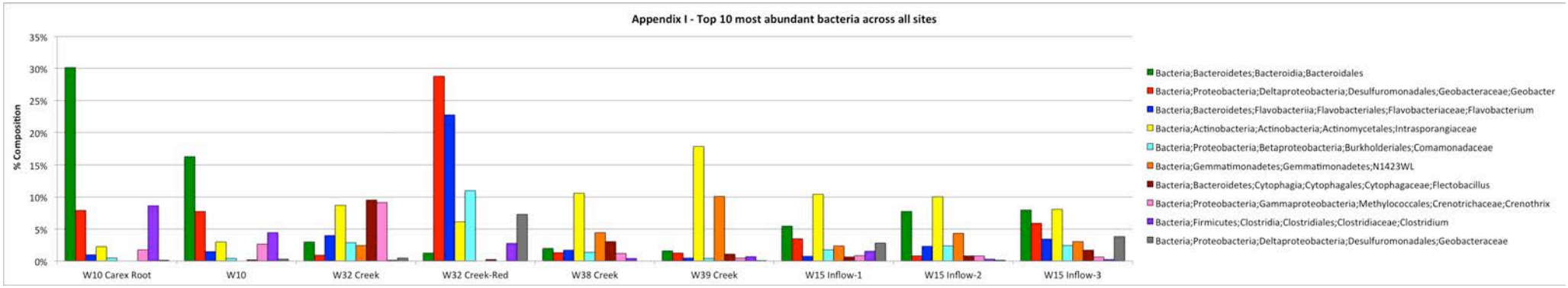


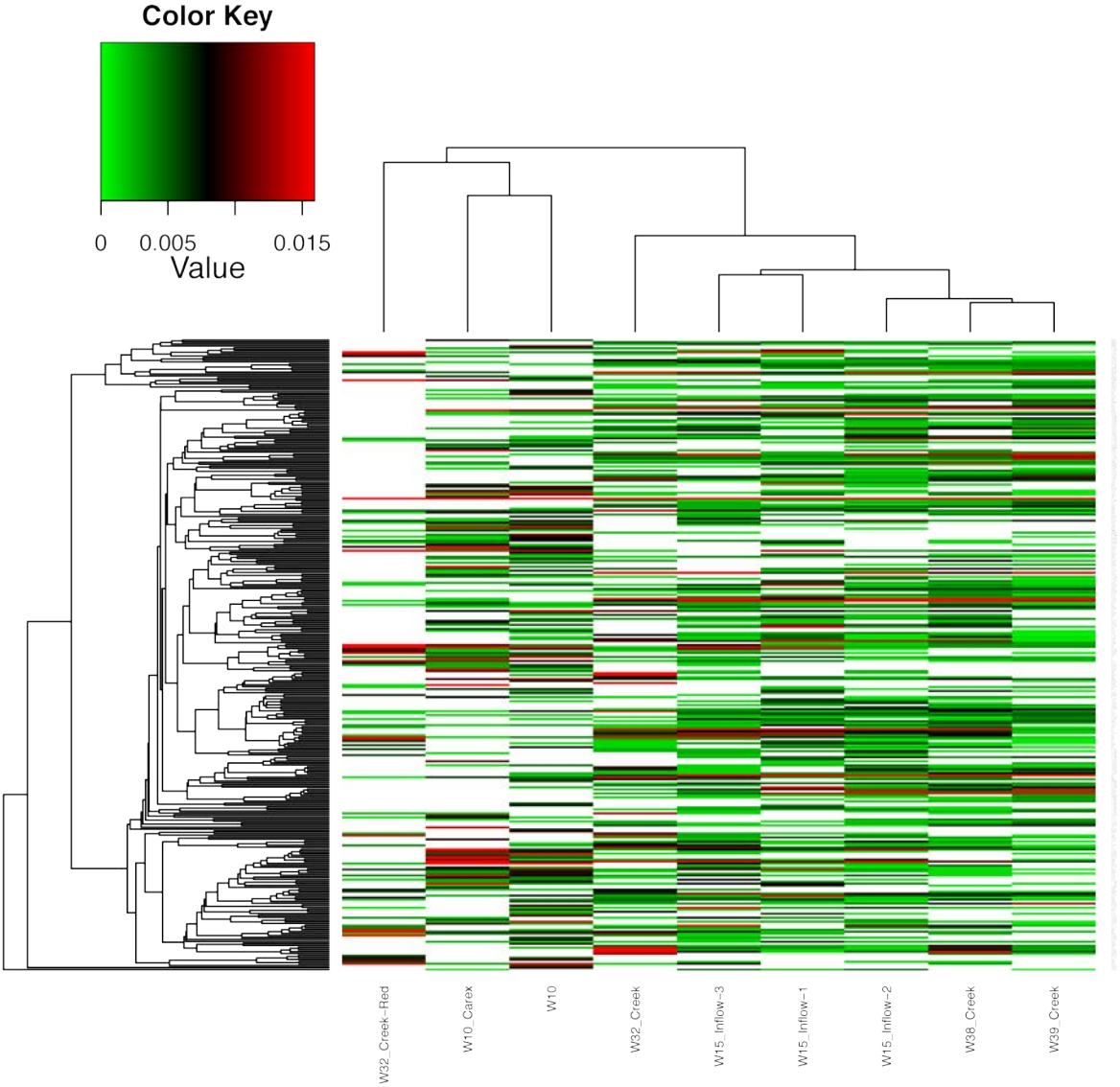
Figure 18: Most-probable number of total heterotrophic organisms at the Minto site. The number of heterotrophic organisms growing under aerobic (with oxygen) and anaerobic (no oxygen) conditions is shown. Significantly more total heterotrophs grew under aerobic conditions that under anaerobic, as demonstrated by a P-value < 0.05.



Figure 19: Color distribution of sediment surrounding *Carex* root at the W10 location.


Appendix I - Top 10 most abundant bacteria across all sites
% Composition
35%
30%
25%
20%
15%
10%
5%
0%
W10 Carex Root
W10
W32 Creek
W32 Creek-Red
W38 Creek
W39 Creek
W15 Inflow-1
W15 Inflow-2
W15 Inflow-3
Bacteria;Bacteroidetes;Bacteroidia;Bacteroidales
Bacteria;Proteobacteria;Deltaproteobacteria;Desulfuromonadales;Geobacteraceae;Geobacter
Bacteria;Bacteroidetes;Flavobacteriia;Flavobacteriales;Flavobacteriaceae;Flavobacterium
Bacteria;Actinobacteria;Actinobacteria;Actinomycetales;Intrasporangiaceae
Bacteria;Proteobacteria;Betaproteobacteria;Burkholderiales;Comamonadaceae
Bacteria;Gemmatimonadetes;Gemmatimonadetes;N1423WL
Bacteria;Bacteroidetes;Cytophagia;Cytophagales;Cytophagaceae;Flectobacillus
Bacteria;Proteobacteria;Gammaproteobacteria;Methylococcales;Crenotrichaceae;Crenothrix
Bacteria;Firmicutes;Clostridia;Clostridiales;Clostridiaceae;Clostridium
Bacteria;Proteobacteria;Deltaproteobacteria;Desulfuromonadales;Geobacteraceae

Appendix II - Microbial Profile



Heatmap showing the distribution of bacteria at the Minto Mine. Each column represents a sampling location, while each row represents an OTU (group of bacteria). The abundance of each group of bacteria is depicted through a heatmap: white indicates absent, and green through red indicates increasing abundance. Samples are clustered together based on the similarity of microbial communities, and indicated by the dendrogram at the top of the heatmap. The dendrogram on the left represents a phylogenetic tree of all OTUs.



Contango STRATEGIES LTD.

Forward looking. Lateral thinking.

Minto Mine Constructed Wetland Treatment Research Program – Pilot Scale

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1. Abbreviations and Definitions

CWTS(s) – Constructed Wetland Treatment System(s).

Dissimilatory reduction – is a process utilized by specific types of bacteria where an electron donor material (e.g. organic carbon) is oxidized, producing energy for the bacteria and reducing another material (e.g. sulphate, or selenium) as part of the microbial metabolism.

DNA (deoxyribonucleic acid) – is the hereditary material in almost all living organisms. This can be used to identify organisms.

DO – Dissolved Oxygen.

Heterotroph – an organism deriving its nutritional requirements from complex organic substances.

Metagenomics – the study of genetic material recovered directly from environmental samples, related to a community rather than an individual organism.

Microbes – microscopic organisms which may be uni- or multi-cellular. This includes algae, bacteria, fungi, viruses, and yeast.

MPN (Most Probable Number) – a statistical value representing the viable population of microbes in a sample through use of dilution and multiple inoculations.

Oxidation – is the loss of electrons or increase in oxidation state by a molecule, atom, or ion. Can be driven by microbes. Process is complementary to chemical reduction.

Redox – Oxidation-reduction potential, measured in millivolts.

Reduction (chemical) – is the gain of electrons or a decrease in oxidation state by a molecule, atom, or ion. Can be catalyzed by microbes. Process is complementary to chemical oxidation.

RRC (Removal Rate Coefficient) - a numerical value, expressed as the variable *k,* representing a first order rate of removal, considering removal extent and the time required for removal.

SeIV broth or agar – laboratory growth medium that contains selenite (SeIV) and indicates reduction of selenite to elemental selenium through a visible colour change.

Species – one of the basic units of biological classification and a taxonomic rank. Rank in the classification of organisms below genus and above strain.

Taxonomy – classification, identification, and naming of organisms.

TSS – Total Suspended Solids.

YTS250 (Yeast, Tryptone, Starch, 250g each) – a type of laboratory growth medium used for environmental microbes.

2. Introduction

Constructed Wetland Treatment Systems (CWTSs) have been proposed as a method for improving the quality of site runoff water in the post-closure period for the Minto Mine (Minto Phase V/VI Expansion Project, YOR Project Number 2013-0100). Once established, wetlands can become self-sustaining ecosystems with the plants providing yearly renewal of carbon to fuel microbial activity. As such, they possess the desirable potential to remediate contaminated mine drainage for as long as it is generated. In order for CWTSs to be effective, however, they must be designed, piloted, optimized, implemented, and maintained in a site-specific manner.

Wetlands provide a natural environment in which unique biogeochemical reactions can occur. CWTSs can be designed to have desirable conditions similar to those in natural wetlands, and should be designed specifically for each site. Having a site-specific approach allows for the appropriate set of operational conditions to be achieved, where complex coupled reactions can take place for treatment of targeted constituents. This can be done with complex and challenging waters, where problematic constituents are transferred or transformed to less bioavailable forms. The benefits of CWTSs include:

- low operational cost,
- low maintenance,
- effective and robust treatment,
- increased effectiveness over time,
- aesthetically pleasing,
- tolerance to changes in flow rate and contaminant load, and
- simultaneous treatment of multiple constituents more effectively than chemical or physical treatment processes.

CWTSs have been used to mitigate risks to aquatic receiving environments from a variety of aqueous contaminants, including copper (Cu) and selenium (Se). The biogeochemical reactions that take place in a wetland environment are many, but there are specific reactions that may decrease the aqueous and more bioavailable concentrations of some contaminants (e.g. Cu, Se), thus lowering risk to the receiving environmental systems by decreasing exposure to contaminants. A process called dissimilatory sulphate reduction can chemically reduce available aqueous sulphate to sulphide through the metabolism of specific bacteria. Once dissimilatory sulphate reduction has occurred, it provides a mechanism to remove dissolved cations such as Cu from water by complexation with the sulphide ion to produce copper sulphide, an insoluble precipitate. Selenium treatment in a CWTS is a similar process, but with slightly different results. The mechanism for Se treatment is more direct, through dissimilatory selenium reduction that is also microbially mediated, often by some of the same bacteria that undergo dissimilatory sulphate reduction. The most common forms of aqueous Se are selenite and selenate, with valence states of IV and VI, respectively. When these are reduced in a wetland to an insoluble state, Se precipitates as elemental Se(0).

In both cases, the Cu and Se precipitates are removed from the water through filtration mechanisms inherent in wetlands and are sequestered into the sediments over time as an accretive process. Accretion is the naturally occurring process of accumulation of wetland

sedimentary material (soil, minerals, decaying plant material, etc.) over time. Once an accreting CWTS is established and mature, targeted constituents are sequestered into the sediment and covered over time by newly generated sediments and detritus. This essentially seals away the treated constituents under layers of sediment, rendering them less bioavailable and less susceptible to re-entry into the water column. There is therefore no need to dredge or harvest wetland plants for the Minto site; in fact, this type of activity would disrupt the treatment functions and re-expose the previously sequestered constituents. This process mimics what occurs in naturally created wetlands; therefore, this process is often the best option for long-term, low-maintenance and effective treatment.

3. Background

At the request of Minto Explorations Ltd, Contango Strategies has undertaken a pilot-scale study to inform the design of site-specific passive treatment technologies for the Minto site during post-closure. These pilot-scale trials build upon a site assessment that was conducted and reported on in 2013. The overriding goal of this pilot study is to develop, optimize, and test site-specific pilot-scale CWTSs for post-closure water treatment at the Minto site.

4. Pilot System Operations

4.1. **Design and assembly**

In October 2013 three pilot CWTSs were constructed, each in duplicate, to evaluate the performances of different designs for selenium and copper treatment (Figures 1 and 2). These were planted with either Sedge (*Carex aquatilis*), or Sedge and aquatic moss as follows:

- System 1 and 2: *Carex aquatilis*
- System 3 and 4: *Carex aquatilis* + Moss
- System 5 and 6: *Carex aquatilis* + Moss, with biochar amendment in soil

The pilot CWTS systems were built with two cells per series, each cell having a 41 cm diameter and a height of 57 cm. To establish the plants, soil was filled to a depth of 20 cm at the bottom of each cell, and submerged under 10 cm of tap water. The water depth was changed throughout the pilot testing to evaluate effects on performance, and develop strategies for optimization.

4.1.1. Hydrosoil

Hydrosoil is a specially selected soil blend to aid in achieving the targeted hydraulic conductivity required to meet the CWTS objectives. In this case, the hydrosoil mixture was the same for all pilots; sand amended with 2% woodchips (v/v) and 5% peat, except that the third system design (system 5 and 6) also included a 5% pine biochar amendment. The sand and amendments were tested for initial metal and nutrient contents (Table 5). This specific type of biochar was chosen as an amendment as previous studies conducted by Contango have demonstrated that addition of biochar to the hydrosoil can result in faster establishment and growth of newly planted wetland plants (SETAC Presentation #593, Haakensen 2013). The biochar selected

was low in soluble salts, had a high surface area, low metals content, and moderate available organic carbon and phosphorous (Appendix 1). The peat borrow site at the Minto Mine was selected as it provided a sufficient quantity for a full-scale wetland. It was tested for metals content and other relevant parameters such as nutrients (N, P, K), and total organic carbon before use in the pilot CWTS system.

4.1.2. Plants

Two plant species were of interest for the pilot CWTS based on data collected during the August/September 2013 site assessments (reports 2013-0100-256 and 2013-0100-257 on the YESAB registry). *Carex aquatilis* (sedge) was chosen, as it is readily available at the Minto site and was one of the first plants to colonize cleared areas, indicating that it may be a good candidate for quickly establishing the CWTS. Moss was also used in the second and third pilot CWTS designs (systems 3-6), as locations at Minto with shallower water had *Carex* and Moss coexisting. As such, we wanted to assess whether the Moss contributes positively or negatively to treatment, and also whether succession from *Carex* to Moss as the dominant plant can be prevented or enhanced through modification of water depth.

It is well known that Moss has a high uptake rate of cations (such as copper and selenium) and is also a relatively benign sink for these elements (i.e., is not a food source for invertebrates or higher animals and as such does not contribute greatly to bioaccumulation (Haines & Renwick, 2009; Longton, 1997; Suren & Winterbourn, 1991)). This was also confirmed in the site assessment in 2013, which produced data indicating that Moss at the Minto site has a high uptake of elements such as copper, and could therefore be an effective means of removing elements from seepage through uptake. During the September 2013 site visit, Moss and *Carex* were collected from the Minto site and these were subsequently used to vegetate all pilot CWTS in this study. A total of 10-12 *Carex* were planted per cell, with approximately a 250ml volume of Moss (compressed) added to systems 3-6.

4.1.3. Water and Testing Periods

The CWTS(s) were designed for treatment of water that would be present at the Minto site after closure. As such, this water does not yet exist and had to be synthesized as an approximation based on water quality modeling. The worst-case post-closure water chemistry was used based on the month of October, as it had the highest concentrations of key elements for any month with free-flowing water. The synthetic water underwent five iterations of design, testing, and optimization to ensure the composition and method of formulating the water resulted in analytical chemistry that matched the water quality predicted to exist after closure and reclamation of the Minto mine. The final recipe and resultant water chemistry is provided in Tables 1 and 2.

Design of the synthetic water chemistry focused on constituents requiring treatment and elements/compounds that are expected to affect treatment rates or capacity based on previous experience and scientific literature. For the first period (High N) of the pilot-phase work, the predicted maximum worst-case ammonia and nitrate concentrations were used, which reflected residuals of these compounds from blasting. The ammonia concentrations measured on site are

lower than would be expected based on the chemical composition of ammonia nitrate used for blasting, suggesting that the ammonia has already undergone oxidation to nitrate prior to the seepage exiting the waste rock pile. Nitrate can affect compounds such as selenium that are removed from water through coupled biogeochemical reactions and dissimilatory reduction, as nitrate has a high affinity for available electrons and is preferentially reduced before other compounds. However, it is known that over time in closure, the concentrations of ammonia and nitrate will decrease dramatically, and this may affect removal rates of other compounds. As such, the pilot-scale CWTSs were subjected to a 6-week trial of synthetic water of the same max worst-case post-closure water quality, but with depleted ammonia and nitrate concentrations to simulate the chemistry expected in long-term closure (Low N period).

It is possible to shift the functionality of a CWTS to stimulate additional microbial dissimilatory selenium reduction and enhance overall treatment of this element. The resultant CWTS would be a hybrid bioreactor-CWTS and is a "passive care" option. This means that the hybrid-CWTS would require more maintenance and operational intervention than a normal CWTS, with periodic dosing of an organic amendment needed (once or a few times annually) in order to achieve the targeted decreased dissolved oxygen and redox levels. However, depending on actual inflow and target outflow concentrations, this approach may be a necessary compromise.

The third and final stage of pilot testing therefore included the addition of an organic carbon amendment (alfalfa hay and straw) to stimulate reducing conditions for improved selenium treatment. Alfalfa hay and straw were chosen as they would provide both short and long-term carbon sources compared to a metered dosing of more available electron donors such as ethanol. This hybrid bioreactor-CWTS period was conducted as an 8-week trial to test how the CWTS functions once converted from conventional CWTS operation to the hybrid configuration. A total of 1.4 g of alfalfa pellets were added per liter of water in each A cell. Straw was added at 10 cm^3 per L to A cells, and 5 cm^3 per L to B cells. The alfalfa pellets and oat straw were tested to ensure suitability for use (Appendix 1). A summary of the periods and timelines is provided in Table 3.

Over the course of the study, a total of 45,000 L of synthetic water (influent) was received by the pilot CWTS (in 41 batches), approximately 1,200 L of synthetic water per week in a pilot CWTS footprint of 1.56 m^2. The system operated with a flow rate of 20 mL/min, resulting in a hydraulic retention time (HRT) ranging from 34 to 94.3 hours, depending on the period of testing and water depth (Table 3).

4.1.4. Routine Monitoring of Explanatory Parameters

Explanatory parameters are characteristics of a specific environment that describe conditions in the context of how these parameters can affect the chemical reactions taking place. These parameters, when within specific ranges of values, can be used to predict and/or promote certain reactions in the aforementioned environment. Accordingly, dissolved oxygen (DO; mg/L and %), temperature (°C), pH, conductivity (µS/cm), specific conductivity (SPC; µS/cm), and oxidation-reduction potential (ORP; mV) were measured on a routine schedule with a YSI Professional Plus handheld unit.

Hydrosoil reduction-oxidation (redox) potential was measured using inert electrodes (copper wire probes with platinum tips) that were permanently installed in the hydrosoil of the cells and remained there for the duration of the project to ensure accurate readings (Faulkner et al., 1989). On testing days, a reference electrode (Accumet Calomel) was suspended in the water column above the inert electrode. The redox potential was then measured in millivolts by a voltmeter.

4.1.5. Flow Rates

Constant flow metering pumps were used to set the flow rate of water into the system. To confirm flow rates, the influent to cell A was collected into a graduated cylinder over a timed period of several minutes allowing a flow rate to be calculated in milliliters per minute (mL/min). Flow rates were measured regularly (at minimum weekly) and adjusted as necessary to within +/-10% of the desired flow, to achieve the desired HRT. The rate of water outflow from a CWTS cell was measured using the same procedure. Outflow rates for a series were typically measured every 2 weeks.

4.1.6. Water sampling

Water samples were collected from the outflow port of each CWTS cell. Care was taken to avoid collecting plant matter, invertebrates, or other debris in the sample. Samples were collected from downstream to upstream to ensure the water sampling did not affect sample collection from the next cell (e.g., disrupting flow rates or suspending solids). All samples were collected according to the schedule in Table 4, with parameters outlined in Table 5.

4.1.7. Stem Counts and Height Measurements

Carex aquatilis stems were counted in each CWTS cell, and 10 stems were randomly chosen in each cell for height measurement. The height of the tallest leaf of each stem was measured to the nearest 0.5 cm from the soil surface to the tip of the leaf. Stem counts and heights were taken monthly during the High N period, as well at the end of the Low N and hybrid bioreactor-CWTS periods (Table 4).

4.1.8. Sediment Sampling

Sediment/hydrosoil samples were collected at key intervals as per Table 4. Sediment was collected by carefully taking a grab sample of a small section of the top 6 cm of hydrosoil in each CWTS cell into a 50 mL centrifuge tube, pouring off any excess water after collection. In addition to the analyses listed in Table 5, sediment was also collected for microbial profiling according to the schedule in Table 4 and as described in Section 4.3.

4.1.9. Plant Sampling

Both *Carex* and Moss were collected for metals analysis at the beginning of the study, as well as at end of the Low N period. This schedule was chosen as it was uncertain if the Moss would survive through the hybrid bioreactor-CWTS period. For sample collection at the end of the Low N period, 4 stems of *Carex* (~15% of total stems) were randomly selected from each cell and trimmed at the surface of the water. An approximate 250 ml volume of compressed Moss was

also collected from each cell in series 3-6. Plants from replicate series (e.g., series 1&2, 3&4, 5&6) were combined and sent for metals analysis as a representative sample for each system type. For the final sampling point (i.e., at pilot system takedown) all remaining *Carex* plants in each cell were trimmed at the soil level and rinsed with water prior to sending for metals analysis.

4.1.10. Freeze-Thaw Trial

In addition to routine analysis for metals in plants, a replicate of each plant sample collected at the end of the low N period was used in a freeze-thaw trial to determine if the plants will leach constituents of concern such as Cu and Se after freezing. In this trial, 200 mL of water from the corresponding CWTS cell was added to the plant sample to completely cover the plant material. The samples were frozen at -20°C for 5 days, and then thawed at room temperature (~22°C) overnight. The plant samples were removed and lightly squeezed to return residual water to the container, and the plant tissue was sent for metals analysis to compare with pre-freeze concentrations.

4.1.11. Organic Carbon Leach Test

To determine the constituents that could leach from the alfalfa and straw that were added for organic carbon during the hybrid bioreactor-CWTS period, a leach test was performed. In this test, 2 L of outflow from system 1 was collected and alfalfa and straw were added to mimic the concentrations that were added to the system. The water was tested for total metals 3 weeks after the organic carbon addition to determine if any metals had leached from the alfalfa and straw into the water.

4.2. **Methods for microbial analyses**

4.2.1. Growth-based analyses (MPN)

The most-probable number (MPN) of bacteria was determined for all micro samples. The MPN test allows for an estimation of the number of bacteria that can grow in a specific laboratory medium. The media tested in this project were specific for bacteria that can reduce nitrate, selenite, and sulfate, respectively, as well as the total number of heterotrophic bacteria for comparative purposes (i.e., a growth-permissive test for all bacteria capable of growing on organic carbon as an energy source). MPN tests for sulphate (Active Standard ASTM D4412) were conducted under anaerobic conditions, whereas selenite (Siddique et al., 2006) and nitrate (Nitrate Reduction Test, supplied by Sigma-Aldrich) were tested under aerobic conditions. Total heterotrophs (grown with YTS250 medium; (Lefrançois et al., 2010) were quantified in both aerobic and anaerobic conditions.

In brief, sediment samples were diluted 1:100 with a 0.1% peptone solution. This starting dilution was then used for a serial dilution along a sterile 96-microwell round-bottom plate containing the respective growth media. Sediment dilutions tested in the MPN plate ranged from 1/400 to 1/419,430,400. At minimum, all tests were conducted in duplicate. Wells were incubated at +30°C without light and assessed for visible growth (formation of a bacterial pellet) and/or colour change specific to the type of media after 27 days. A colour change to black or red/orange indicated sulphate or selenite reduction, respectively. Nitrate reduction was tested

as per the kit manufacturer's protocol by addition of reagents to differentiate between reduction to nitrite or nitrogen gas. The most probable number of bacteria capable of each metabolism was then calculated as described in the FDA Bacteriological Analytical Manual, Appendix 2: Most Probable Number from Serial Dilutions: (http://www.fda.gov/Food/FoodScienceResearch/LaboratoryMethods/ucm109656.htm). The MPN of organisms in each pilot system was compared over time using the non-parametric Kruskal-Wallis test.

4.2.2. DNA-based analyses

DNA was extracted from all samples using the MO BIO Powersoil Powerlyzer DNA extraction kit, with the addition of phenol during cell lysis as per the manufacturer's alternative protocol. Targeted sequencing was used to identify bacteria present in each sample via polymerase chain reaction (PCR) amplification of the v3/v4 region of the 16S ribosomal RNA gene (Klindworth et al., 2013). Library preparation and sequencing was performed as per the manufacturer's instructions for MiSeq v3 paired-end 300 bp sequencing (Illumina). All raw sequences were filtered to remove low quality reads, and the forward and reverse primers were truncated. Bioinformatics pipelines consisting of internal scripts and selected QIIME scripts (Caporaso et al., 2010; Edgar, 2010) were used to process the reads. Similar sequences were clustered into groups called Operational Taxonomic Units (OTUs) using a 97% identity threshold. Taxonomic classification of the OTU's was performed according to the Greengenes database (DeSantis et al., 2006; McDonald et al., 2012). Non-metric multidimensional-scaling was used to compare microbial communities across samples using the Bray-Curtis dissimilarity metric. The percentage of the microbial community classified as organisms of interest (e.g., sulphate-reducing bacteria) was also compared across samples. The most abundant OTU's were compared to the Cu, Se, and S concentrations in the sediment to calculate the Spearman correlation coefficient and associated p-value.

5. Results and Discussion

5.1.1. Outflow concentrations of copper and selenium

The pilot systems were operated for 37 weeks with three distinct trial periods of High Nitrogen, Low Nitrogen, and Hybrid Bioreactor-CWTS, with the average influent and outflow water chemistry provided in Tables 6, 7, and 8, respectively. The minimum, maximum, and average measurements of selected parameters and copper and selenium concentrations are summarized in Table 9. While copper treatment was relatively stable through both the High N and Low N periods, selenium treatment improved during the High N period (Figure 3), and then stabilized through the Low N period. For both the High N and Low N periods, the systems that were planted with Moss and *Carex* had significantly better copper treatment than Moss only. There was no statistically significant difference found among the three pilot system designs for selenium removal (i.e., Moss or biochar did not show a significant effect on selenium treatment during these periods).

The hybrid bioreactor-CWTS configuration was effective for both copper and selenium treatment. The purpose of the bioreactor-CWTS configuration is to serve as a contingency in

case greater treatment is needed than is being achieved by the CWTS at a given point in time (e.g., the influent water quality or outflow objectives change). The tests here show that the CWTS can be repurposed as a hybrid bioreactor-CWTS with improved selenium and copper treatment within the same footprint, and without need for added construction. However, operating the system as a hybrid bioreactor-CWTS would require greater operational maintenance than it would in its original CWTS configuration.

In the case of selenium, the reconfiguration resulted in a statistically significant decrease in outflow concentration compared to that prior to the reconfiguration (Figure 3). This result for selenium was expected based on the design targeting more favourable conditions for selenium reduction, such as lower dissolved oxygen and ORP resulting from the additional electron donors from the microbial decomposition of the alfalfa and straw. In terms of copper, the lowest outflow concentration of the study was measured during the hybrid bioreactor-CWTS phase (7.5µg/L); however, prior to this there was a brief decrease in treatment effectiveness in all systems (Figure 3, June 19th data point). However, even during this brief disruption, 86% of the total copper entering the system was treated.

It is not known whether this brief change in treatment effectiveness was due to a short-term loss of treatment capacity during the transition, or if it might have been caused by a release of copper and other constituents that were sorbed to the moss or other organic matter during the transition. This decrease in treatment effectiveness coincided with low dissolved oxygen and moderately reducing conditions in the water column being attained in all cells (DO < 0.5 mg/L, ORP < -80 mV); however, it was not clear if the additional copper in the outflow was released from the alfalfa and straw that were added, or if constituents were briefly released from the sediment or moss and detritus after the system transitioned into more reducing conditions. As a result, a leach test was conducted on the alfalfa and straw and determined that there was no discernible release of copper from these organic materials. The origins of the decrease in treatment effectiveness could not be determined definitively, with the disturbance being registered during a single sampling point, with a maximum potential disturbance period of 4 weeks.

While the hybrid bioreactor-CWTS configuration improved selenium treatment, and demonstrated the lowest outflow concentration of copper of the entire trial, the short upset of treatment performance during the transition is meaningful in terms of monitoring during site implementation. It should be emphasized that if a transition from a typical CWTS design to a hybrid bioreactor-CWTS style is performed, careful monitoring of the CWTS is required as some elements may have decreased treatment effectiveness for a brief period. During this timeframe, if concentrations of elements exceed regulatory guidelines, water may need to be recycled through the wetland, or a contingency water treatment plan should be put in place for the outflow over the period of treatment disturbance.

5.1.2. Removal Rate Coefficients

An important factor for wetland design is the rate of treatment, also known as the removal rate coefficient (RRC). The RRC is based on treatability of a specific compound and the hydraulic retention time of the system. Once the RRC is known, the equation is rearranged and the RRC can be used to calculate the HRT needed to meet outflow objectives (and accordingly, the size of the wetland). It can also be used to predict the expected outflow concentrations of a specific element in a wetland of the same design based on inflow rates, concentrations, and footprint. RRCs vary with the concentrations of an element, and therefore, a sequential cell approach was used in this pilot study to allow for the RRC to be calculated over a broader range of inflow and outflow concentrations than if the system were only considered as a single treatment unit. Together these can then be used to estimate the size needed for a full-scale CWTS to meet outflow concentration performance objectives for a given influent concentration and flow rate. It should be noted that a RRC is specific to each given constituent respective to a particular wetland design, and can change based on the water chemistry and/or system design. The RRCs presented in this paper are a representative approximation of the Minto Mine site in a closure scenario and accounts for High N, Low N, and the use of a hybrid bioreactor-CWTS operational period (Figure 4, Table 10). When comparing the RRC of different designs of pilot systems in which the retention time and inflow concentrations are the same, a higher RRC indicates increased removal of the constituent. In the equation below, the first order RRC has been reconfigured to calculate *t* (hydraulic retention time), allowing the use of *Ci* (initial concentration), *Cf* (final, desired concentration), and *k* (removal rate constant, RRC) to be used to size the wetlands accordingly.

$$k = \frac{-ln(C_f/C_i)}{t}$$

$$t = \frac{-ln(C_f/C_i)}{k}$$

Equation: Removal rate coefficient (RRC) calculation, and rearranged to solve for HRT

When comparing the copper RRC of cells A and B within a series during the High N period, the RRC of the first cell is very high, while the RRC of the second cell is negligible (Figures 5 and 6). This phenomenon is frequently observed in treatment wetlands, where the treatment rate is faster at higher concentrations than at lower concentrations (Horner et al., 2012; Murray-Gulde et al., 2008; Spacil et al., 2011). As such, increasing the HRT of these systems by adding additional cells to the series would not result in a linear decrease in the final copper outflow concentration.

In contrast, the Low N and hybrid bioreactor-CWTS periods do not have this same degree of difference between copper RRC between cells A and B within a series (Figure 5). This change in cell-to-cell difference of RRC still results in similar or improved outflow concentrations. Generally speaking, the RRC of the first cell in the system has decreased, while that in the second cell has improved. This observation can be attributed to a variety of factors. Based on previous experience, the most likely explanation is that there was an abundance of sorption sites available in all cells at pilot initiation. As the study progressed, the sorption sites became

occupied and removal via this mechanism was no longer available. This same phenomenon is also often seen when setting up an on-site demonstration-scale or full-scale treatment wetland. As such, both pilot and demonstration scale studies should be conducted for a sufficiently long period of time so as to realize the actual treatment capacity of the system beyond this initial sorption phase. This phenomenon must also be taken in the context of the duration over which the system must operate within a year, and the knowledge that the reaction front will move downstream within a wetland through this time. Therefore, an increased wetland size might not significantly improve outflow concentrations, but will aid in maintaining treatment performance through the year, so long as the added size does not result in concentration of the elements due to evaporation of the water caused by increased size.

A second possibility is that since the High N conditions coincided with an algal bloom (both visible, but also confirmed by community profiling that there was an abundance of cyanobacteria; Figure 7), the greater concentrations of cyanobacteria and algae during the High N period could have also been responsible for increased sorption and uptake of copper in the A cells. In either case, it is not to say that the B cells were not capable of a high removal rate, but rather that the concentration of copper in the A cell effluent left very little copper available for treatment in the B cell, regardless of capacity for treatment.

It is important to note that the RRCs for copper decreased during the hybrid bioreactor-CWTS period (Figure 4). This can be attributed to the longer HRT created by the increased water depth in this period despite similar outflow concentrations. Through all periods, the outflow concentration of copper for any given system did not vary significantly, indicating that regardless of the cause of the difference in RRC at different points in the system operations (e.g., sorption, algal uptake, increased water depth), the thermodynamic minimum for copper removal has likely been approached within the system, with an average outflow concentration of 18 ± 5 µg/L (Figure 6). That is to say, making the wetland bigger using any of the tested designs would not result in a further significant decrease in outflow copper concentration, but rather, design alterations would need to be made in order to attempt to achieve lower outflow concentrations. During the High and Low N periods when the Moss was actively growing, copper removal rates were greater in systems containing Moss than in systems with only *Carex*. It is likely that as the Moss continues to grow over time and increase in total mass, the RRC would also improve in these systems. Finally, the addition of biochar did not have any significant effect on performance.

Selenium tells a different story than copper in regards to the RRC. Selenium had a fairly similar removal rate in both the A and B cells during the Low N period (Figures 5 and 6). This could be because the selenium treatment is microbially mediated, or because the inflow concentrations are already so low that both the inflow and outflow concentrations of all cells still fall within the same RRC range. A test of the metals composition in the plants and subsequent mass balance of the systems confirmed that the system was operating as designed to target selenium mineralization to the sediment, as plant uptake was responsible for only 1-3% of the total selenium that was removed from the water during the entire course of the study (Figure 10). Based on the RRCs for selenium in the Low and High N periods, an increased HRT is expected

to result in a lower outflow concentration of selenium (Figure 6). This could be achieved by increased footprint, or water depth, so long as the depth does not exceed that in which the *Carex* and Moss will grow.

Conversely, the hybrid bioreactor-CWTS time period had a significantly higher RRC for selenium in the A cells compared to the B cells (Figure 5). It should be noted that while the HRT is longer during the hybrid bioreactor-CWTS period than it was during the Low or High N periods due to an increase in water depth, it can be calculated using the RRC from the previous periods that this alone does not account for the degree of improvement in outflow concentration of the bioreactor-CWTS design. If the improvement in outflow concentration could be attributed solely to the increased HRT, the RRC would remain the same. Instead, we observe an improvement of not just the outflow concentration, but also in improvement in the RRC for selenium in all systems when converted to the bioreactor-CWTS configuration. This higher RRC is presumably due to the more favourable conditions that have been targeted for selenium reduction, including lower dissolved oxygen and a lower ORP resulting from the additional electron donors available through the microbial decomposition of alfalfa and straw. Through all time periods, there was no discernible benefit measured by the addition of biochar into the soils of the systems; rather, the systems that were planted with *Carex* and Moss performed similarly regardless of whether biochar had been included in the soil or not.

As would be expected based on the RRC data for systems containing *Carex* and Moss, the final sediment concentrations of copper are higher in the A cells than the B cells, with the B cells being only slightly higher in concentration than the starting levels (Figure 8). The same trend is evident for selenium; however, the starting concentration and concentrations in most of the B cells are below detection limits. It is of note that the copper and selenium sediment concentrations are positively correlated to an increase of sulphur in the sediment, while no discernible change was found in iron concentrations (Figure 8). This correlation is important to understand the long-term stability of copper within the system, as copper is highly stable when mineralized in a sulphide form and maintained under reducing conditions such as those in the wetland soils. In this case, the sulphides are being formed by microbes living in the soil and associated with the plant roots (see black areas in soil and around roots in Figures 14 and 15). This is supported by the finding of an initial increase of sulphate-reducing bacteria in the sediment, followed by stability over time at a most probable number of approximately 10,000-100,000 microbes per gram of sediment that are capable of reducing sulphate to sulphide (Figure 17). In the case of selenium, the correlation with sulphur in sediment is presumably due to the ranges of explanatory parameters found within the system (dissolved oxygen, ORP, soil redox, pH) which are favorable to both microbially-mediated selenium and sulphur reduction, suggesting the sedimentary selenium is likely in an elemental form, or sorbed to organic materials such as biofilms. While, the stability of the copper and selenium mineralizations were not tested in this study, it is planned for testing to be performed during the upcoming on-site demonstration-scale trial by using sequential extractions on the sediment paired with ICP analysis.

5.1.3. Fate and distribution of copper and selenium

Mass balances were calculated for each pilot system to assess the fate and distribution of elements removed from the water during the course of testing (i.e., the full 37 week period). For systems that were planted with *Carex* and Moss, 64-75% of the copper removed from the water was sequestered into the top 6 cm of sediment (Figure 9). In contrast, systems with only *Carex* planted and no Moss had only 33% of the copper ending up in this top sediment fraction (Figure 9). For systems planted with both *Carex* and Moss, 68-83% of the treated selenium was sequestered to this top portion of the sediment, but less than 52% was in the sediment of the system planted with only *Carex* (actual percentage cannot be calculated as the selenium was below the sediment detection level; Figure 10). For both elements, only 1-3% of the total load removed from the water was found in the above water vegetation (*Carex*) at the end of pilot testing.

It was not possible to determine the concentrations of elements in Moss, as the majority did not survive in the hybrid bioreactor-CWTS period conditions. However, in anticipation of this potential issue, samples of Moss were sent for analysis prior to the start of the hybrid bioreactor-CWTS period, allowing for a bioconcentration factor (BCF) to be calculated.

5.1.4. Plant uptake and overall performance

The bioconcentration factor (BCF) is the ratio of the amount of an element taken up into or sorbed onto the surface of a plant, compared to the amount of the element in the water, and is expressed as the ratio of mg of chemical per kg of organism to mg of chemical per liter of water. The amount of an element sorbed onto the surface of a plant was included in the BCF calculation to give a better representation of the amount that would be present in and on a plant in the wetland (i.e., for plant death/degradation as well as consumption perspectives). As would be expected based on samples collected during the site visit in 2013 and supported by scientific literature, we found in this study, and also in the site visit that *Carex* had a much lower BCF than Moss for both selenium and copper (Table 11). This finding is expected, as moss is known to have a high sorption capacity and therefore, high BCF (Aldrich & Feng, 2000; Gstoettner & Fisher, 1997). For some elements, bioconcentration factors within a plant species are higher when the element itself is at a lower concentration. This held true in the pilot systems when comparing cells A and B in a series, as the BCF for copper in *Carex* was significantly higher in B cells (which had lower aqueous concentrations) than in A cells (with higher influent concentrations) for all three pilot system designs tested (Figure 11). However, for selenium, the A cells had a similar BCF as B cells planted with both *Carex* and Moss (not statistically different), but a significantly higher BCF in the A cells planted with only *Carex* (Figure 11).

Selenium bioconcentration in *Carex* in the pilot systems was very low, resulting in average final tissue concentrations of only ~1.9 mg/kg wet weight. The BCF of selenium was greater in the Moss than the *Carex* in the pilot systems (Table 11). However the Moss BCF recorded here is still relatively low overall based on the United States Environmental Protection Agency Toxic Substances Control Act which considers BCF less than 1000 as non-bioaccumulative. It should also be noted that although selenium is generally known to bioaccumulate, Moss is not a food source for higher animals or invertebrates; as such, the selenium in the Moss should not be

contributing to bioaccumulating pathways (Haines & Renwick, 2009; Longton, 1997; Suren & Winterbourn, 1991).

It is important to the overall treatment system design to determine if elements may be released by the plants by freezing and thawing in the water, as would occur through natural seasonal progression. A freeze-thaw trial was conducted in which *Carex* and Moss were collected and sent for metals analysis both before and after being submerged in water from the wetland and freezing to -20°C followed by thawing to room temperature. It was promising to find that *Carex* released only a small percentage of the selenium (18-38%) and copper (0-13%) as noted by the decrease in tissue concentrations after freezing and thawing (Figure 12). Moreover, Moss was found to have a significant increase in concentration of these elements after being frozen and thawed (Figure 13). It appears that through the freeze-thaw process, the Moss has an enhanced sorption of these compounds, removing an additional amount from the water.

Based on the pilot system design and plant densities, *Carex* would theoretically release on average 0.14 mg of copper and 0.27 mg of selenium per square meter of wetland area, while Moss would theoretically bind 511 and 11.8 mg/m^2, respectively. This finding is important, as it demonstrates that despite some release of constituents by *Carex* through freeze-thaw, the Moss will provide uptake in excess of what is released by the *Carex* and also assist with treatment during initial spring thaw before biogeochemical processes are at full capacity.

During the pilot CWTS takedown, it was noted that the *Carex* had an extensive root structure throughout the entire hydrosoil (Figures 14 and 15), making it an ideal candidate for construction of stable wetlands on site. Additionally, it should be noted that no significant difference in plant health or growth (i.e., number of stems, plant height, weight, and percent moisture at end of pilot) was found for systems with or without biochar or Moss (Figure 16). Based on the data collected during the pilot-phase testing, we can conclude that the combination of *Carex* and Moss is the ideal CWTS design for long-term and sustainable copper and selenium treatment. In this design the *Carex* provides structure to soil and sediment of the wetland with extensive root structure, additional treatment capacity based on the roots drawing water down into the sediment, renews organic carbon in the sediment and contributes to accretion through detritus decomposition, and aids in the production of sulphate reducing zones (as indicated by black areas in Figure 15). Meanwhile, the Moss provides sorption and uptake of elements even through freeze-thaw events.

5.1.5. Microbially-mediated treatment

In order to fully understand the function of biogeochemical processes such as the mineralization of copper and selenium, it is critical that the microbial populations performing these functions are well characterized and understood. Therefore, the microbial communities populating the sediment of each pilot CWTS cell were analyzed by both genetic and growth-based methods. The genetic method identifies and comparatively quantifies all of the bacteria in a sample without requiring growth, which is very important because a large proportion of environmental organisms will not grow in a laboratory setting. The genetic-based profiling is therefore used to assess which microbes are present and at what abundance, giving an idea of the robustness of

the community and allowing for potential function to be inferred. The second method is called the most-probable number (MPN) assay, which is a dilution-based growth method that can determine the number of selenite-reducing, sulphate-reducing, and nitrate-reducing organisms per gram of sediment. The MPN method is therefore a means to evaluate the biogeochemical cycling potential of a sample; however, because the analysis is based on growth in laboratory media, it is presumed that not all organisms are able to grow and that only a partial profile of the microbial community is provided. Both methods allow for comparisons to be made of the microbial populations across samples, treatments, or time series.

With the microbial analyses, we were interested in three key questions:

- How does the microbial community adapt and change with different influent water chemistries and system designs?
- Are microbes that contribute to selenium, copper, or nitrate treatment present in the pilot wetland cells, and if so, are these same organisms present at the Minto site?
- Is selenium or copper treatment mediated by microbes present in the sediment?

As Figure 7 shows, the microbial population adapted rapidly and consistently to the changes in influent chemistry (i.e., High N, Low N), as well as the addition of organic matter. The nMDS plot of Figure 7 can be interpreted simply as samples that cluster together have more similar microbial communities, while those that are farther apart have distinct differences in their populations (such as identity or relative abundance). At the beginning of the pilot system operation, the microbial community had a proportionately high number of *Actinobacteria*, which then transitioned to *Cyanobacteria* during the High N period corresponding to a visible algal bloom. The Low N and hybrid bioreactor-CWTS phases had similar microbial populations, indicating that either the microbes present in the Low N phases were also well suited to the bioreactor-CWTS conditions, or that the community was still in flux at the final sampling, and treatment would have continued to improve over time in the hybrid system as more anaerobic organisms flourished. Overall, Figure 7 shows that as expected, the microbial populations were both adaptive and robust, shifting to respond to changes in the water chemistry and explanatory parameters such as dissolved oxygen and redox.

The results of the MPN testing indicate that the High N period had an abundance of organisms capable of denitrification including the reduction of nitrate to nitrite, and nitrate to N_2 gas (Figure 17). As expected based on the elevated water concentrations of nitrate, the High N period had a large number of organisms present that could reduce nitrate to N_2 gas. Although in decreased abundance, denitrifying bacteria were present through the Low N period and the hybrid bioreactor-CWTS period, despite having much lower nitrogen concentrations during this timeframe. As such, it is anticipated that fluctuations in ammonia and nitrate concentrations over time would be rapidly met with millions of microbes per gram of soil that are able to perform denitrification reactions.

Sulphate-reducing bacteria consistently increased in abundance over time in all system designs, starting out with a most probable number of approximately a thousand per gram of sediment and increasing by two orders of magnitude through the study (Figure 17). This increase is

consistent with trends observed in the soil redox measurements becoming more negative over time in this study. The presence and abundance of sulphate-reducing bacteria were positively correlated with final sulphur concentrations in sediment, according to both the genetic and growth-based profiling methods. Several groups of known sulphate-reducing bacteria (e.g., *Desulfobacterium, Desulfosarcina, Desulfotomaculum)* were found to also correlate with copper concentrations in the sediment, suggesting that their presence leads to reduction of sulphate to sulphide, which then binds to copper and precipitates into the sediment. This can also be demonstrated by the black areas surrounding the *Carex* roots and black precipitates on the sediment surface observed during system takedown (Figures 14 and 15). This copper-sulphide mineralization mechanism is ideal for treatment as it provides a non-bioavailable and stable form of copper in the sediment.

We can tell from the genetic sequencing that these same types of sulphate-reducing bacteria were found in samples that were collected during the August 2013 site visit, specifically associated with *Carex* roots and sediment at the W10 location, as well as in the sediment at W15. It is likely that these were brought to the pilot system along with the plant roots as these *Carex* were harvested from the Minto W10 area. This indicates that despite having different sediment and environmental conditions at the site in comparison to the pilot CWTS, these microbes are robust and can be further encouraged to thrive and perform beneficial functions by having the ecological niches that they prefer (e.g., *Carex* roots, low ORP and DO).

The most probable number of selenite-reducing bacteria ranged from hundreds of thousands to tens of millions per gram of soil and was similar between the different system designs, having similar variance through the three testing periods (Figure 17). There was a general increase in selenite-reducing bacteria during the High N and Low N periods, followed by a decrease in their numbers during the hybrid bioreactor-CWTS period similar to denitrifying organisms discussed above. It was not possible to calculate a correlation between the number of selenite-reducing bacteria and the selenium concentration in the sediment due to the generally low levels (i.e., non-detectable) of selenium in the sediment of most cells. An abundance of selenite-reducing organisms in the pilot systems as well as at the Minto site indicates that these organisms are a robust and diverse group, and are present with a latent capacity to perform microbially-mediated selenium treatment given the proper conditions (e.g., electron donors, targeted ORP and DO levels).

The importance of these microbial results is two-fold: first, the treatment of selenium, copper, or nitrate can be mediated and sustained in a very passive manner by microbes if provided the right environment. Second, this necessary environment can be targeted through system design and adjustments in operation as have been done through this study, resulting in stimulation of these microbes in abundance, diversity, and activity.

5.1.6. Range of operational parameters

In order to be able to effectively implement a full-scale passive treatment system such as a CWTS, the targeted explanatory parameters must be predefined. As such, one of the goals of this pilot-scale work was to confirm or optimize the ranges of these parameters from what can

be derived from publications and previous experience. These explanatory parameters can then be used to predict the functionality of a wetland and efficacy of treatment. Moreover, well-defined explanatory parameters allow for adaptive management plans to be developed in order to maintain optimal treatment effectiveness and predictability. The pilot systems tested here performed well not only across a variety of water chemistries and induced conditions, but also over a wide range of water pH, DO, ORP and soil redox. Table 9 includes the minimum, maximum and average recorded values of some selected explanatory parameters for the different periods of testing. Most importantly, this table shows that the range of conditions needed for effective copper and selenium treatment were achieved for this site-specific CWTS design, highlighting the robustness and adaptability of the design and providing a range of conditions under which the system will function.

6. Conclusions and Recommendations

When designed and implemented in a strategic and scientifically guided manner, CWTSs can mitigate risks posed by many contaminants. A treatment plan that includes processes that precipitate insoluble species of these constituents for sequestration into the sediments of the wetland are very desirable, as this mechanism captures the contaminant and stores it with stability in the sediments, rather than transferring the contaminant to an indeterminate fate (e.g., plant uptake or volatilization). This study has addressed several important design considerations regarding implementation of a CWTS at the full scale for the treatment of copper and selenium at the Minto Mine. A synopsis of the conclusions and recommendations are:

- An effective and robust pilot-scale CWTS design was demonstrated and optimized using plants from the Minto site and predicted closure water chemistries.

- The pilot-scale CWTSs provided much of the information needed to effectively size full-scale CWTS for treatment at the Minto site, but sizing will also be dependent on other factors (hydrology, available area, and constructability considerations).

- The recommended CWTS design for long-term passive treatment of copper and selenium at the Minto site is a combination of *Carex* and Moss planted in a sand substrate supplemented with woodchips and peat.

- The selected design achieved average outflow concentration of 12 µg Cu/L (average influent 146 µg Cu/L), and 6 µg Se/L (influent 10.2 µg Se/L).

- Conversion of the CWTS to a hybrid bioreactor-CWTS design resulted in the lowest copper (7.5 µg/L) and selenium (1.9 µg/L) concentrations of the study, but was preceded by a brief disruption of treatment capacity. This configuration would require greater maintenance than the basic CWTS, but could be implemented if any changes in influent water quality are predicted or if outflow objectives change.

- If transitioning from the CWTS design to hybrid bioreactor-CWTS, close monitoring of CWTS and potential consideration of contingency treatment is recommended during the

transition period, as there may be a brief and temporary disruption in treatment effectiveness.

- The majority of the copper and selenium treated by the system was sequestered to the top 6 cm of sediment, with only a small portion (1-3%) being taken up into the *Carex* (commonly known as sedge).

- After undergoing a freeze-thaw cycle, the *Carex* used in this study released only a small amount of copper and selenium; however, under the same trial the binding and sorption capacity of the Moss increased, resulting in a net removal of copper and selenium from the water through a freeze-thaw cycle.

- Fluctuations in nitrogen levels (as would be expected in early closure due to residue from blasting activities) are responded to by a diverse microbial community with a robust capacity to treat nitrate, but also continue to treat Cu and Se through this transition;

- There were thousands to billions of nitrate, sulphate, and selenium reducing bacteria present per gram of soil in the pilot CWTS, many of which were confirmed through genetic testing to be the same groups of organisms as found at the Minto site.

- It is recommended that the design be tested on-site at Minto with a demonstration-scale CWTS receiving seepage water.

7. Closure

We trust the information herein satisfies your present requirements. Should you have any questions, please contact the persons listed below. We appreciate the opportunity to provide the services detailed in this report, and look forward to discussing any comments you may have.

Respectfully submitted,

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9. Tables

Table 1: Synthetic water formulation.

Element	Compound	Concentration (mg/L)
Aluminum	$Al_2(SO_4)_3 \cdot 18H2O$	0.929
Ammonium	NH_4HCO_3	1.00[1]
Arsenic	$Na_2HAsO_4 \cdot 7H_2O$	0.012
Barium	$BaCl_2 \cdot 2H_2O$	0.817
Cadmium	$CdCl_2$	0.0005
Calcium	$CaCl_2$	11.68
	$Ca(NO_3)_2 \cdot 4H_2O$	197.59[1]
Chromium	CrO_3	0.013
Cobalt	$CoCl_2 \cdot 6H_2O$	0.017
Copper	$CuCl_2 \cdot 2H_2O$	0.239
Iron	$FeSO_4 \cdot 7H_2O$	15.29
Lead	$Pb(NO_3)_2$	0.0007
Magnesium	$MgSO_4 \cdot 7H_2O$	324.77
Manganese	$MnSO_4 \cdot H_2O$	8.10
Molybdenum	$Na_2MoO_4 \cdot 2H_2O$	0.026
Nickel	$NiCl_2 \cdot 6H_2O$	0.023
Potassium	KNO_3	28.34[1]
	KH_2PO_4	0.298
Selenium	$Na_2SeO_3 \cdot 5H_2O$	0.032
Silver	$AgNO_3$	0.0005
Sodium	Na_2SO_4	54.14
Strontium	$SrCl_2 \cdot 6H_2O$	8.73
Vanadium	V_2O_5	0.026
Zinc	$ZnCl_2$	0.056

[1] During Low N and hybrid bioreactor-CWTS periods of testing, NH_4HCO_3 and KNO_3 were not added, and the $Ca(NO_3)_2 \cdot 4H_2O$ was decreased to 1.976 mg/L

Table 2: Influent chemistry for pilot CWTS.

Element (total)	unit	Influent				
		Targeted	Acclimation Period	High N	Low N	Hybrid bioreactor-CWTS
Aluminum (Al)	mg/L	0.100	0.089	0.085	0.080	0.098
Antimony (Sb)	mg/L	0.00160	0.00060	0.00065	0.00061	0.0006
Arsenic (As)	mg/L	0.0030	0.0031	0.0033	0.0033	0.0037
Barium (Ba)	mg/L	0.500	0.153	0.284	0.192	0.224
Beryllium (Be)	mg/L	0.00078	<0.001	<0.001	<0.001	<0.001
Boron (B)	mg/L	0.315	0.0243	0.0262	0.0239	0.0243
Cadmium (Cd)	µg/L	0.3340	0.011	0.236	0.336	0.392
Calcium (Ca)	mg/L	82	70.33	73.90	40.73	48.22
Chromium (Cr)	mg/L	0.0065	0.0058	0.0064	0.0071	0.0075
Cobalt (Co)	mg/L	0.0039	0.0039	0.0042	0.0043	0.0044
Copper (Cu)	mg/L	0.13	0.14	0.15	0.15	0.18
Iron (Fe)	mg/L	3.1	3.1	3.1	3.2	3.5
Lead (Pb)	mg/L	0.00070	0.00044	0.00067	0.00059	0.00071
Magnesium (Mg)	mg/L	50	49	49	51	53
Manganese (Mn)	mg/L	2.6	2.5	2.5	2.6	2.7
Molybdenum (Mo)	mg/L	0.012	0.013	0.015	0.013	0.013
Nickel (Ni)	mg/L	0.0072	0.0077	0.0087	0.0076	0.0080
Phosphorus (P)	mg/L	0.082-0.27 [1]	0.333	0.236	0.082	0.082
Potassium (K)	mg/L	14	14.3	14.6	3	3
Selenium (Se)	mg/L	0.010	0.010	0.011	0.010	0.011
Sodium (Na)	mg/L	44	40	41	42	43
Strontium (Sr)	mg/L	2.9	2.5	2.5	2.5	2.6
Sulphur (S)	mg/L	85	85	84	86	88
Uranium (U)	mg/L	0.0051	0.0012	0.0012	0.0012	0.0014
Vanadium (V)	mg/L	0.016	0.015	0.013	0.014	0.018
Zinc (Zn)	mg/L	0.032	0.0517	0.037	0.040	0.042
Nutrients						
Total Ammonia (N)	mg/L	0.2-0.5 [2]	0.61	0.50	0.29	0.05
Dissolved Nitrate (N)	mg/L	0.8-28 [2]	29.33	29.67	0.86	0.84
Misc. Inorganics						
pH	pH	7.5-8.0	7.89	8.16	7.96	8.01
Anions						
Alkalinity (Total as CaCO3)	mg/L		107.33	107.00	100	100
Bicarbonate (HCO3)	mg/L		133.33	130.00	122	130

[1] Phosphorous was not modeled for closure, so the maximum concentration from W15 (2007-2012) was used (0.27mg/L), this was later decreased to 0.082mg/L (95th percentile).
[2] Ammonia and nitrate were not added to the synthetic water for the Low N and hybrid bioreactor-CWTS periods.

Table 3: Pilot CWTS testing periods.

Period	Timeline	# Weeks	Water Depth	HRT (hr) ± Std. Dev.	Description/Purpose
Pre-Influent	Nov 4, 2013 – Dec 12, 2013	6	10 cm	N.D.	Plant establishment
Acclimation	Dec 13, 2013 – Feb 26, 2014	10	10cm	37± 3	Acclimation of system to influent
High N	Feb 27, 2014 – Apr 12, 2014	7	10 cm, 17 cm[1]	37 ± 3 53 ± 3[1]	High ammonia and nitrate in influent (early closure)
Low N	Apr 13, 2014 – May 28, 2014	7	17 cm	53 ± 2.5	Low ammonia and nitrate (long-term closure)
hybrid bioreactor-CWTS	May 29, 2014 – Jul 16, 2014	7	30 cm	87 ± 3	Post alfalfa and straw addition (hybrid bioreactor-CWTS)

[1] 17 cm from March 13 to Apr 12, 2014 to test higher water depth during High N Period. This therefore results in a longer HRT for this timeframe.

Table 4: Monitoring schedule.

Water	
Temperature	Weekly
pH	
Dissolved oxygen	
Regulated metals water package [1]	Bi-weekly
Phosphorus	
Ammonia [2]	
Nitrate	
Flow rate	
Alkalinity [3]	Monthly
Hardness [3]	
Conductivity [3]	
Chemical oxygen demand (COD) [4]	
Total organic carbon [4]	
Total Kjeldahl Nitrogen (TKN) [4]	
Sulphate [3]	
Biological oxygen demand (BOD) [3]	Monthly (outflow only)
Total suspended solids [2]	
Soil	
Eh (redox)	weekly
Available NPK	Initial soil, beginning of influent, end of study
Regulated metals package	
Total organic carbon	Initial soil, end of study
Particle size analysis	
Conductivity	
Cation exchange capacity (CEC)	
Sodium adsorption ratio	
Plant	
Regulated metals	Start and end; Freeze-thaw trial
Stem counts and heights [4]	Monthly
Microbial	
Most probable number (growth-based)	Initial soil; beginning High N period; end Low N period; end of study
Genetic microbial community profiles	

[1] Dissolved metals was tested in addition to total metals for the Low N and hybrid bioreactor-CWTS periods
[2] Not performed during hybrid bioreactor-CWTS period, except at takedown of pilot
[3] Bi-weekly during hybrid bioreactor-CWTS period
[4] Not performed during Low N and hybrid bioreactor-CWTS periods, except at takedown of pilot

Table 5: Summary of water, soil, and plant analyses.

Water	Soil	Plant
ICP-MS Total or Dissolved (Al, Sb, As, Ba, Be, B, Cd, Cr, Co, Cu, Fe, Pb, Li, Mn, Mo, Ni, P, Se, Si, Ag, Sr, Tl, Sn, Ti, U, V, Zn, Ca, Mg, K, Na, S)	ICP-MS (Al, B， Sb, As, Ba, Be, Cd, Ca, Cr, Co, Cu, Fe, Pb, Li, Mg, Mn, Mo, Ni, P, K, Se, S, Ag, Na, Sr, Tl, Sn, U, V, Zn,)	ICP-MS (Al, Sb, As, Ba, Be, Bi, B, Cd, Ca, Cr, Co, Cu, Fe, Pb, Mg, Mn, Mo, Ni, P, K, Se, Ag, Na, Sr, Tl, Sn, Ti, U, V, Zn)
Routine Water (Alkalinity, HCO_3, Ion Balance, Dissolved ions, Hardness, Nitrate/Nitrite)	Available Nitrogen, Phosphorous, and Potassium (NPK)	
Total Suspended Solids (TSS)	Cation Exchange Capacity (CEC)	
Chemical Oxygen Demand (COD)	Sodium Adsorption Ratio (SAR)	
Biological Oxygen Demand (BOD)	Total Organic Carbon (TOC)	
Total Kjeldahl Nitrogen (TKN)	Electrical Conductivity (EC)	
Total Organic Carbon (TOC)	Texture Analysis (Physical properties)	
Ammonia		

Table 6: Performance of pilot CWTS during High N period.

Elements (Total)	Unit	Influent	Outflow Average		
		Average	*Carex* only	*Carex* + *Sphagnum*	*Carex* + *Sphagnum* + Biochar
Aluminum (Al)	mg/L	0.085	0.012	0.008	0.009
Arsenic (As)	mg/L	0.0033	0.0010	0.0008	0.0007
Barium (Ba)	mg/L	0.284	0.039	0.036	0.040
Boron (B)	mg/L	0.0262	0.025	0.027	0.055
Cadmium (Cd)	µg/L	0.236	0.038	0.023	0.017
Calcium (Ca)	mg/L	73.9	78	78	77
Chromium (Cr)	mg/L	0.0064	0.0024	0.0023	0.0024
Cobalt (Co)	mg/L	0.0042	0.0008	0.0004	0.0003
Copper (Cu)	mg/L	0.150	0.018	0.013	0.012
Iron (Fe)	mg/L	3.10	0.08	0.08	0.08
Magnesium (Mg)	mg/L	49	50	51	51
Manganese (Mn)	mg/L	2.5	1.1	0.37	0.2
Molybdenum (Mo)	mg/L	0.0150	0.0139	0.0130	0.0128
Nickel (Ni)	mg/L	0.0087	0.0042	0.0027	0.0020
Phosphorus (P)	mg/L	0.236	<0.1	<0.1	<0.1
Potassium (K)	mg/L	14.6	14.9	15.3	15.1
Selenium (Se)	mg/L	0.0110	0.0076	0.0073	0.0067
Sodium (Na)	mg/L	41	43	45	42
Strontium (Sr)	mg/L	2.5	2.7	2.7	2.6
Sulphur (S)	mg/L	84	86	87	86
Uranium (U)	mg/L	0.0012	0.0011	0.0011	0.0011
Vanadium (V)	mg/L	0.0130	0.0023	0.0019	0.0017
Zinc (Zn)	mg/L	0.0370	0.0042	0.0044	0.0034
Nutrients					
Total Ammonia (N)	mg/L	0.50	0.05	0.05	0.05
Dissolved Nitrate (N)	mg/L	29.67	29.4	29.4	29.5
Explanatory Parameters					
pH	pH	8.16	7.47	7.61	7.80
DO (mg/L)			5.33	5.16	5.52
Water redox (mV)			154.5	144.2	136.0
Soil Redox (mV)			-148	-191	-220

Table 7: Performance of pilot CWTS during Low N period.

Elements (Total)	Unit	Influent	Outflow Average		
		Average	*Carex* only	*Carex* + Moss	*Carex* + Moss + Biochar
Aluminum (Al)	mg/L	0.080	0.008	0.006	0.007
Arsenic (As)	mg/L	0.0033	0.0007	0.0006	0.0006
Barium (Ba)	mg/L	0.192	0.028	0.027	0.030
Boron (B)	mg/L	0.024	0.023	0.024	0.022
Cadmium (Cd)	µg/L	0.336	0.070	0.027	0.023
Calcium (Ca)	mg/L	41	47	47	47
Chromium (Cr)	mg/L	0.0071	0.0034	0.0029	0.0030
Cobalt (Co)	mg/L	0.0043	0.0006	<0.0003	<0.0003
Copper (Cu)	mg/L	0.150	0.0210	0.0113	0.0125
Copper (Cu) Dissolved	mg/L	0.037	0.0123	0.0086	0.0086
Iron (Fe)	mg/L	3.20	0.16	0.06	0.09
Magnesium (Mg)	mg/L	51	55	55	55
Manganese (Mn)	mg/L	2.6	0.9	0.2	0.2
Molybdenum (Mo)	mg/L	0.0130	0.0110	0.0104	0.0105
Nickel (Ni)	mg/L	0.0076	0.0035	0.0018	0.0015
Phosphorus (P)	mg/L	0.11	<0.10	<0.10	<0.10
Potassium (K)	mg/L	3.0	3.2	3.4	3.4
Selenium (Se)	mg/L	0.0100	0.0061	0.0060	0.0063
Selenium (Se) Dissolved	mg/L	0.0080	0.0052	0.0053	0.0052
Sodium (Na)	mg/L	42	44	44	45
Strontium (Sr)	mg/L	2.5	2.7	2.7	2.7
Sulphur (S)	mg/L	86	93	94	93
Uranium (U)	mg/L	0.0012	0.0009	0.0010	0.0010
Vanadium (V)	mg/L	0.0140	0.0023	0.0016	0.0022
Zinc (Zn)	mg/L	0.0400	0.0082	0.0032	0.0031
Nutrients					
Total Ammonia (N)	mg/L	0.29	<0.05	<0.05	<0.05
Dissolved Nitrate (N)	mg/L	0.86	0.26	0.25	0.25
Explanatory Parameters					
pH	pH	7.96	7.58	7.76	7.61
DO (mg/L)			5.69	6.05	5.42
Water redox (mV)			179	170	205
Soil Redox (mV)			-148	-183	-209

Table 8: Performance of pilot CWTS during hybrid bioreactor-CWTS period.

Elements (Total)	Unit	Influent	Outflow Average [1]		
		Average	*Carex* only	*Carex* + Moss	*Carex* + Moss + Biochar
Aluminum (Al)	mg/L	0.098	0.010	0.013	0.009
Arsenic (As)	mg/L	0.0037	0.001	0.001	0.001
Barium (Ba)	mg/L	0.224	0.061	0.061	0.058
Boron (B)	mg/L	0.024	0.026	0.026	0.025
Cadmium (Cd)	µg/L	0.392	0.057	0.067	0.067
Calcium (Ca)	mg/L	48	49	48	48
Chromium (Cr)	mg/L	0.0075	<0.001	<0.001	<0.001
Cobalt (Co)	mg/L	0.0044	0.003	0.002	0.002
Copper (Cu)	mg/L	0.180	0.021	0.023	0.026
Copper (Cu) Dissolved	mg/L	0.0350	0.0122	0.0134	0.0168
Iron (Fe)	mg/L	3.50	0.41	0.33	0.22
Magnesium (Mg)	mg/L	53	54	54	55
Manganese (Mn)	mg/L	2.7	5.2	5.1	5.7
Molybdenum (Mo)	mg/L	0.013	0.015	0.020	0.021
Nickel (Ni)	mg/L	0.008	0.006	0.006	0.007
Phosphorus (P)	mg/L	0.12	0.11	0.11	0.10
Potassium (K)	mg/L	3.0	6.4	6.1	6.3
Selenium (Se)	mg/L	0.0100	0.0040	0.0040	0.0040
Selenium (Se)	mg/L	0.0119	0.0038	0.0039	0.0041
Sodium (Na)	mg/L	43	44	44	44
Strontium (Sr)	mg/L	2.6	2.6	2.6	2.6
Sulphur (S)	mg/L	88	87	87	87
Uranium (U)	mg/L	0.0014	0.0010	0.0010	0.0010
Vanadium (V)	mg/L	0.018	0.002	0.002	<0.001
Zinc (Zn)	mg/L	0.042	0.007	0.005	0.006
Nutrients					
Total Ammonia (N)	mg/L	0.05	0.06	0.05	0.05
Dissolved Nitrate (N)	mg/L	0.84	0.08	0.06	0.03
Explanatory Parameters					
pH	pH	8.01	7.16	7.20	7.19
DO (mg/L)			1.52	1.73	1.86
Water redox (mV)			-34	-91	-134
Soil Redox (mV)			-197	-218	-226

[1] The average outflow for the hybrid bioreactor-CWTS period includes data from June 19th, when there was a spike in several constituents when the system reached low dissolved oxygen and moderately reducing conditions in the water column.

Table 9: Summary of explanatory parameters and outflow concentrations of copper and selenium

Period	Parameter	Influent	System								
			Carex			*Carex* + Moss			*Carex* + Moss + Biochar		
		Avg	Min	Max	Avg	Min	Max	Avg	Min	Max	Avg
High Nitrogen	Copper (µg/L)	146.4	16	20	18	11	16	13	9	14	12
	Selenium (µg/L)	10.2	5.8	9.7	7.6	6.0	9.5	7.3	5.8	8.5	6.7
	pH	8.00	6.99	7.94	7.49	7.19	8.29	7.64	7.39	8.28	7.84
	DO (mg/L)	-	4.08	6.69	5.36	2.59	7.55	5.35	3.50	8.70	5.75
	ORP (mV)	-	82	215	157	104	215	247	84	205	140
	Redox (soil,mV)	-	-229	-51	-144	-273	-177	-190	-301	-135	-218
Low Nitrogen	Copper (µg/L)	145.5	17	32	21	10	13	11.3	11	15	12.5
	Selenium (µg/L)	10.2	5.3	6.7	6.1	5.6	6.5	6.0	5.8	6.9	6.3
	pH	7.96	7.16	8.05	7.54	7.26	8.32	7.71	7.15	8.25	7.60
	DO (mg/L)	-	3.99	7.72	5.61	3.02	9.78	5.73	2.57	9.04	5.21
	ORP (mV)	-	126	220	168	113	188	159	106	374	182
	Redox (soil,mV)	-	-249	-39.5	-150	-246	-88	-185	-264	-123	-211
Hybrid Bioreactor-CWTS	Copper (µg/L)	175.6	7.5	45	21	8.3	44	23	8.3	64	26
	Selenium (µg/L)	11.9	1.9	5.6	4.0	2.5	5.4	4.0	2.8	5.4	4.0
	pH	8.01	6.91	7.62	7.16	6.95	7.82	7.20	6.96	7.71	7.19
	DO (mg/L)	-	0.02	3.23	1.52	0.11	5.61	1.73	0.03	4.71	1.86
	ORP (mV)	-	-175	-79	-129	-212	-101	-150	-195	-30	-134
	Redox (soil,mV)	-	-232	-170	-197	-268	-159	-218	-290	-150	-226

Table 10: Removal rate coefficients for each system and period of testing.

Element	System	Period RRC (hr^{-1})		
		High N	Low N	hybrid bioreactor-CWTS [1]
Copper	*Carex*	0.0465	0.0390	0.0250
	Carex + Moss	0.0519	0.0497	0.0236
	Carex + Moss + Biochar	0.0543	0.0462	0.0242
Selenium	*Carex*	0.0069	0.0103	0.0130
	Carex + Moss	0.0078	0.0102	0.0128
	Carex + Moss + Biochar	0.0095	0.0090	0.0123

[1] The calculation of RRC for the hybrid bioreactor-CWTS period included data from June 19th, when there was a spike in several constituents when the system reached low dissolved oxygen and moderately reducing conditions in the water column.

Table 11: Bioconcentration factor for each plant species at the end of the Low N period.

Plant Species	Copper (L/kg)	Selenium (L/kg)
Carex	23	113
Moss	1390	593

10. Figures

Figure 1: Pilot CWTS layout

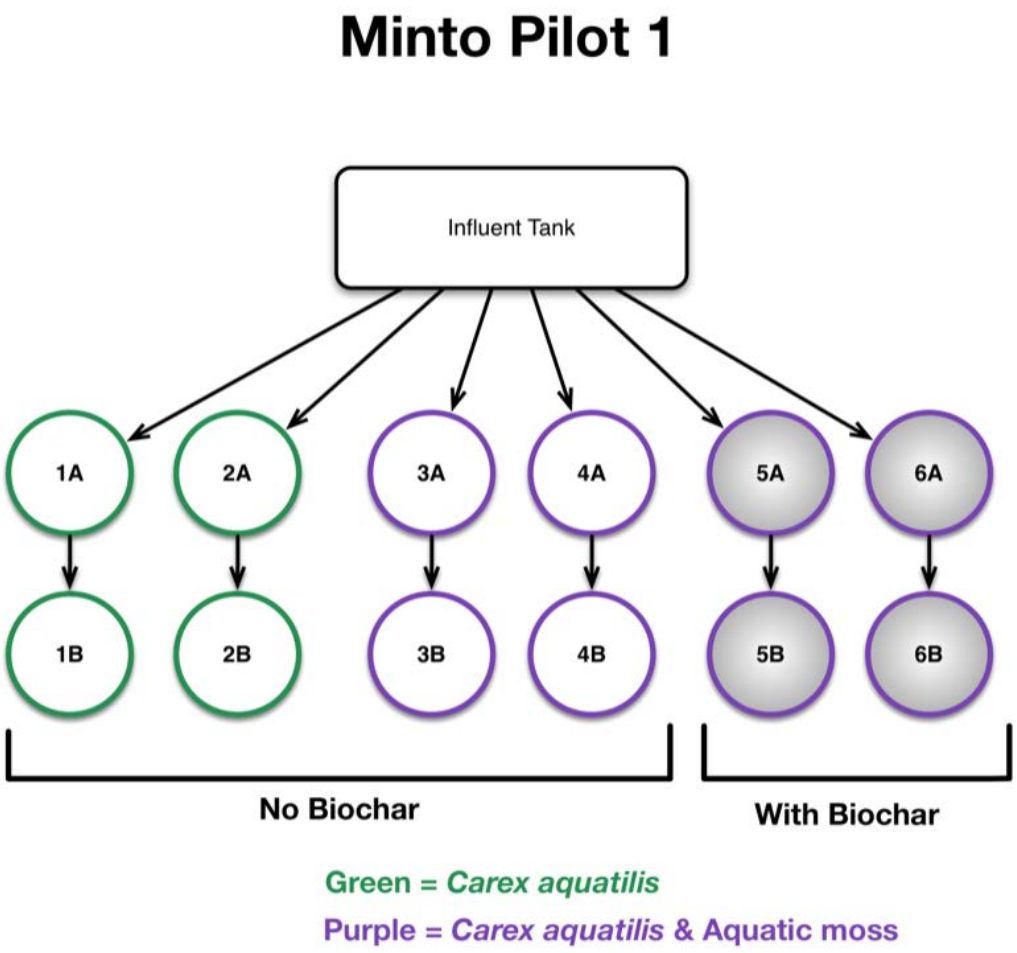


Figure 2: Photographs of pilot CWTS set up. Top is entire system, with 6 series of two cells; bottom is one cell.



Figure 3: Outflow copper and selenium water concentrations over time compared to influent.

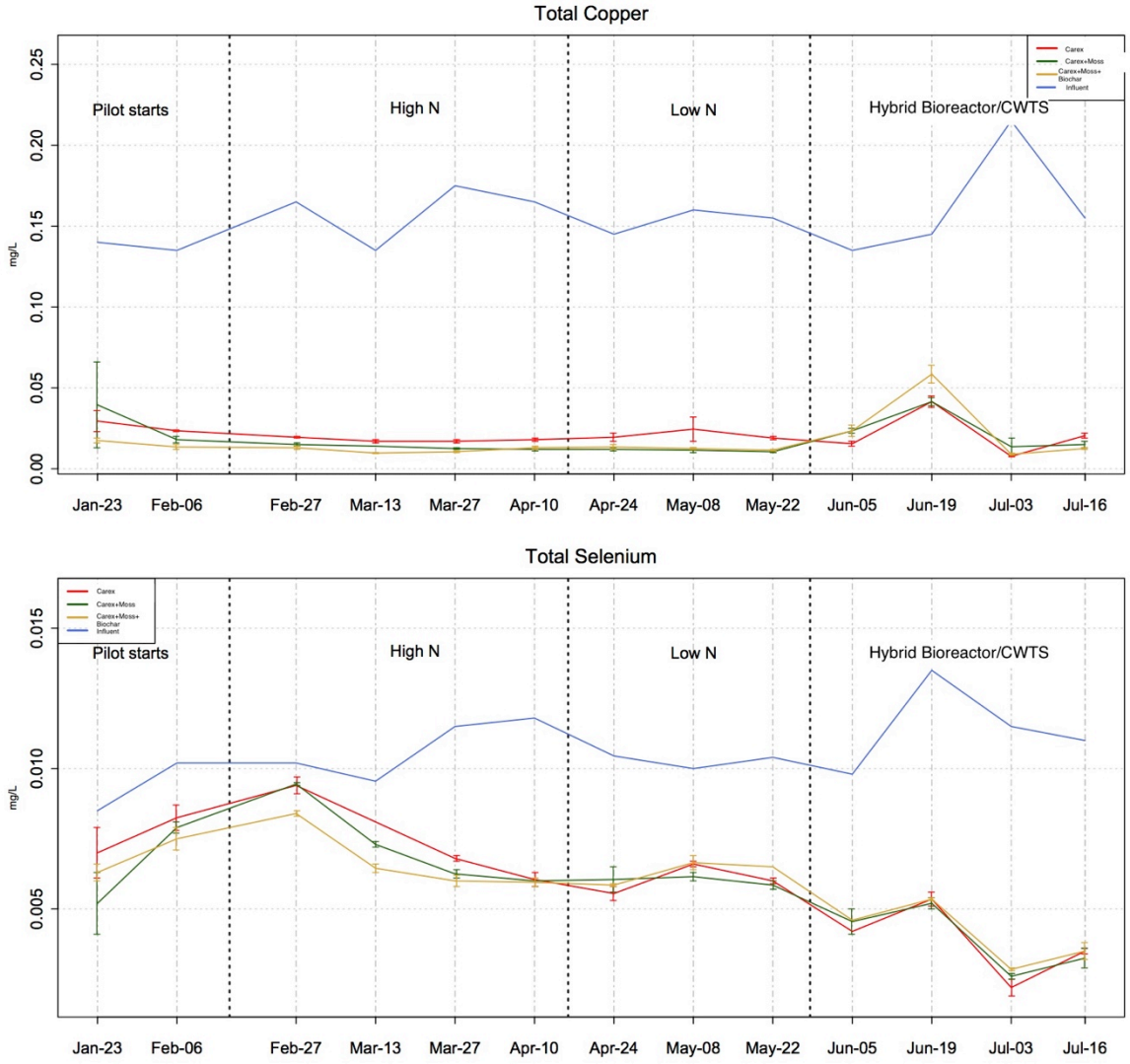


Figure 4: Copper and selenium removal rate coefficients during all periods of testing for the three system designs.

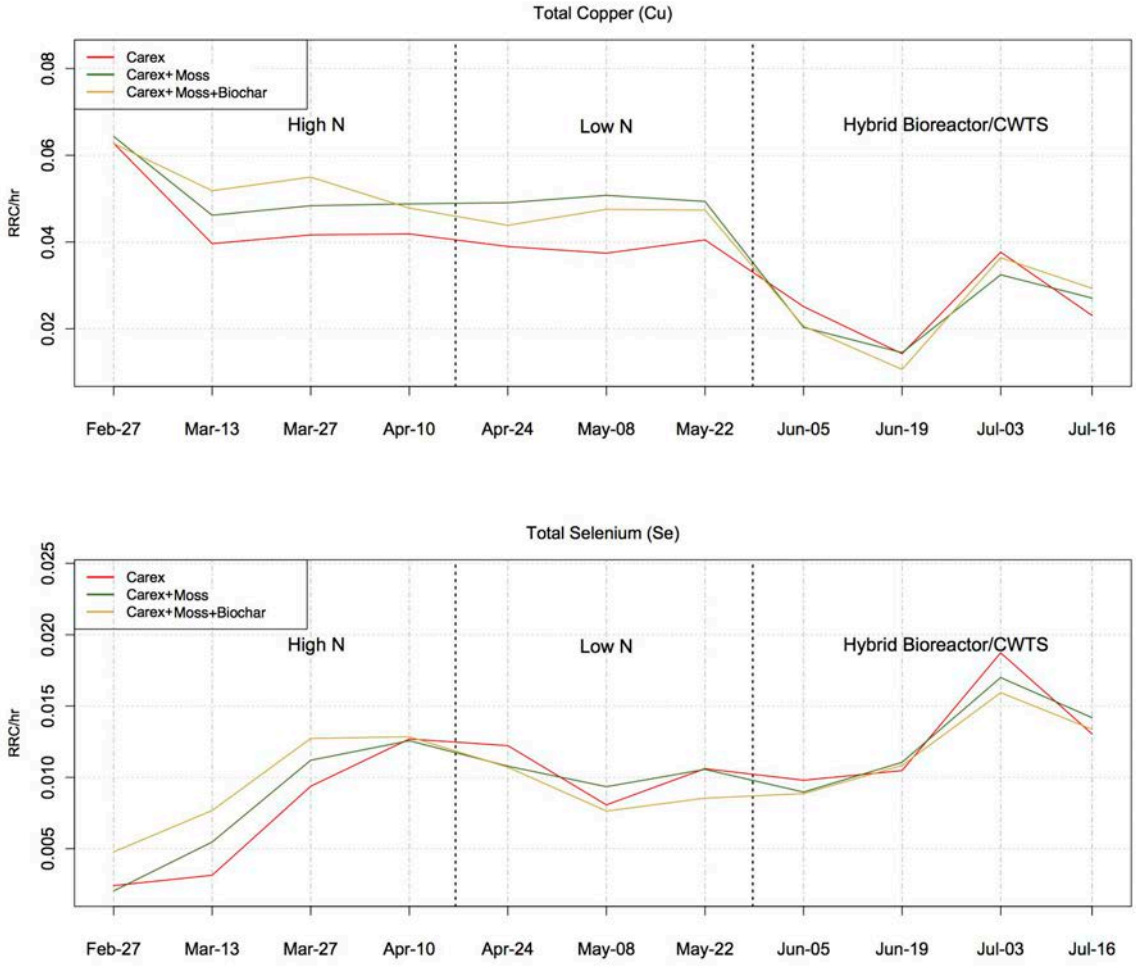


Figure 5: Copper (top) and selenium (bottom) removal rate coefficients for cells A and B in each system design. All three periods of testing are plotted separately (High N = Feb 27-April 12; Low N = April 13-May 28; hybrid bioreactor/CWTS = May 29-July 16). The average RRC is indicated with the bar chart, and error bars show the range. Note that the water depth was changed during the hybrid bioreactor/CWTS phase, which impacts the RRC.

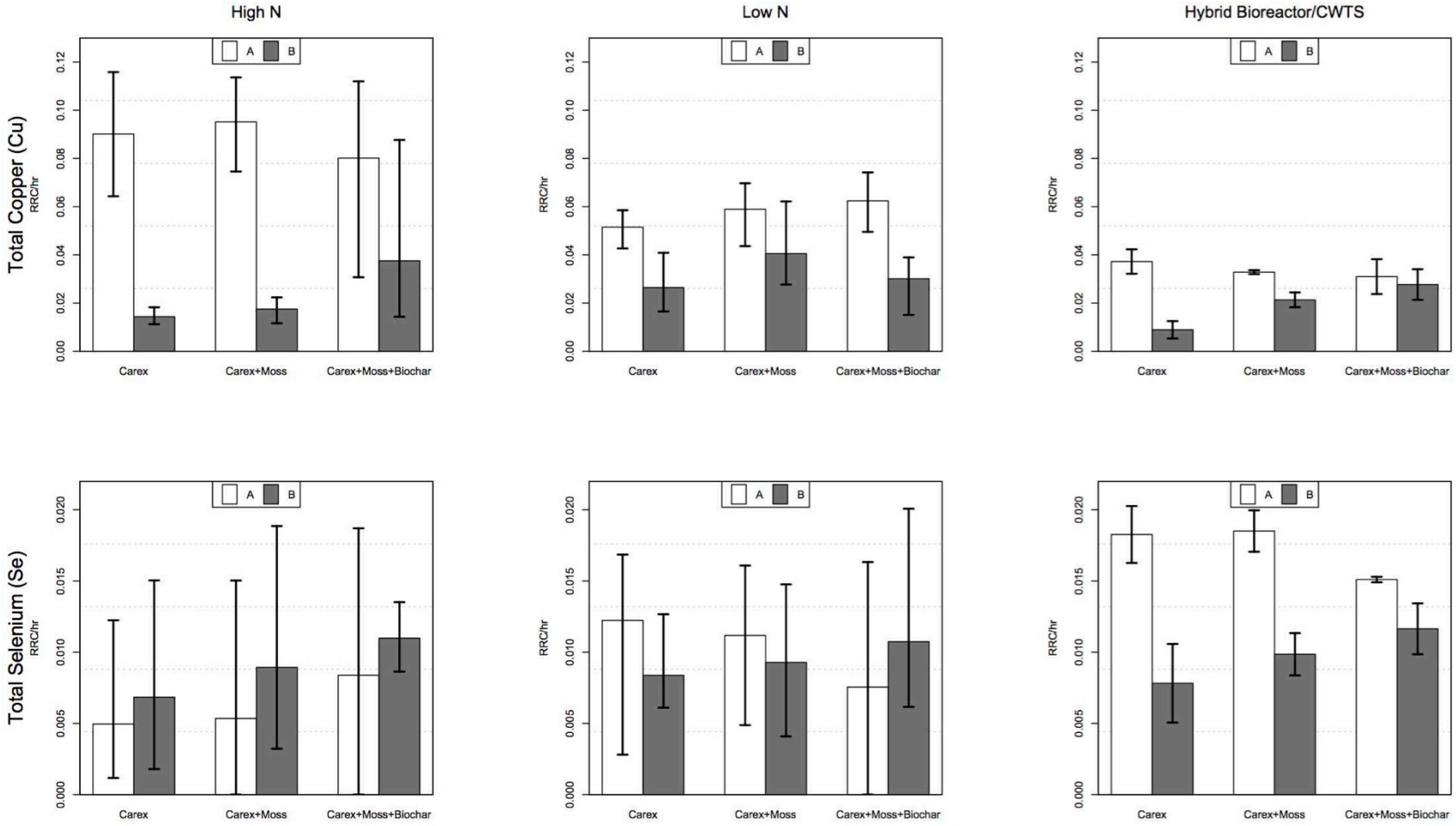


Figure 6: Outflow copper and selenium water concentrations from A and B cells over time compared to influent.

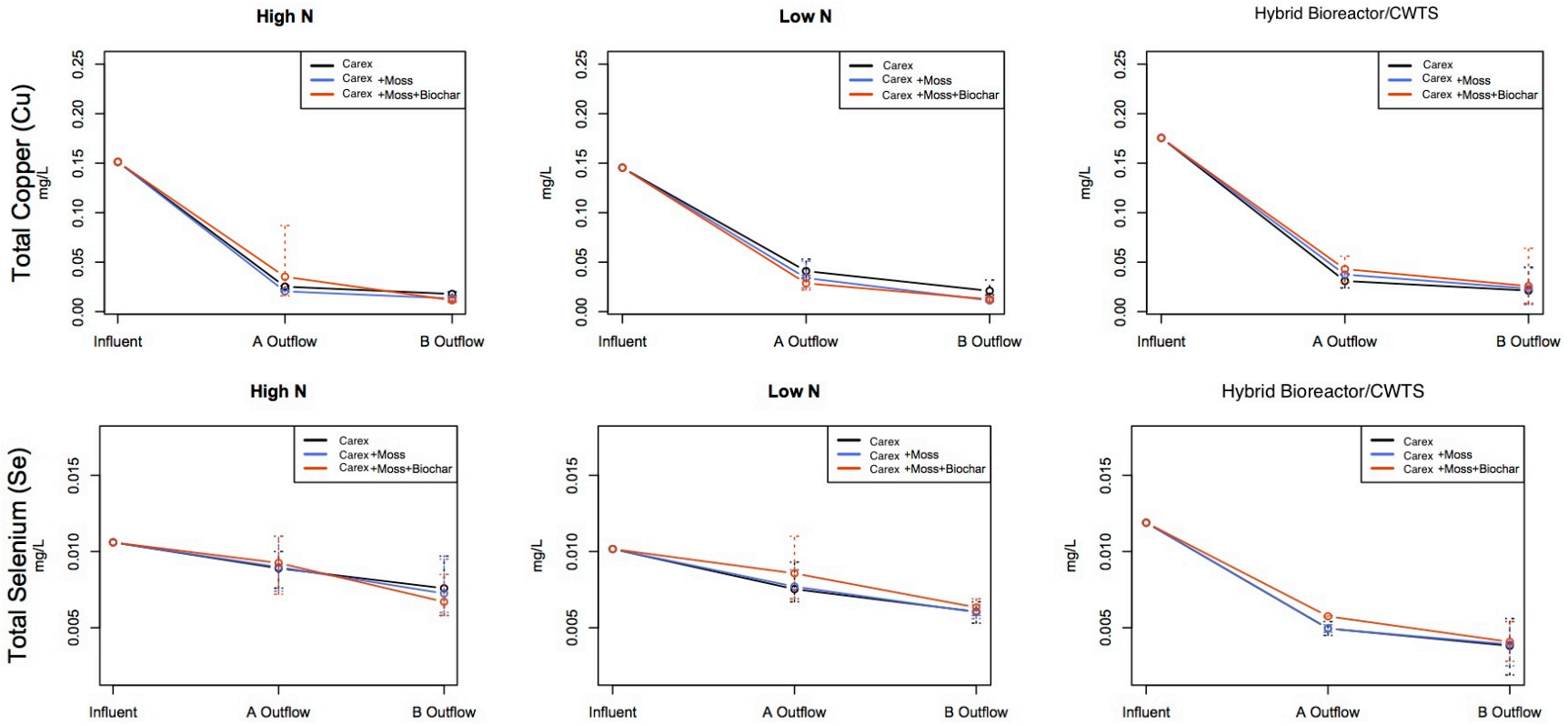


Figure 7: Non-metric multidimensional scaling (nMDS) analysis plot for bacterial community profiles of each pilot system design over time. Results are based on the genetic profiling method. Each time point during testing is shown with a different color, while the system design is indicated with shapes. All points that cluster together indicate similar bacterial communities. At the end of the pilot testing, we also determined the microbial population of the top root mass (orange outline), which was very similar to the microbial population in the sediment. The top 10 bacterial groups are indicated on the plot with dark blue numbers. The distance between a bacterial group label and a sample point depicts how high that group is in that sample.

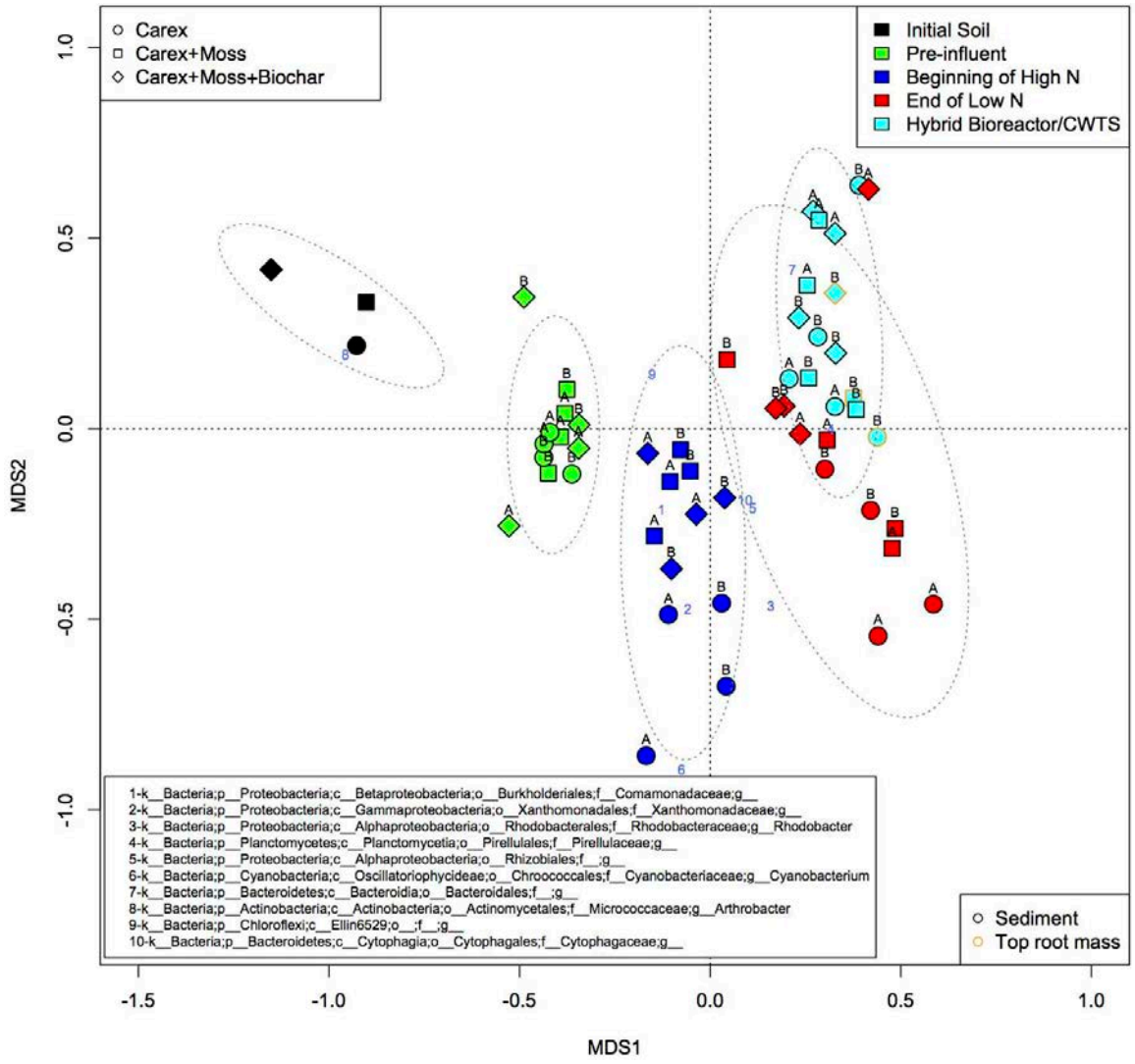
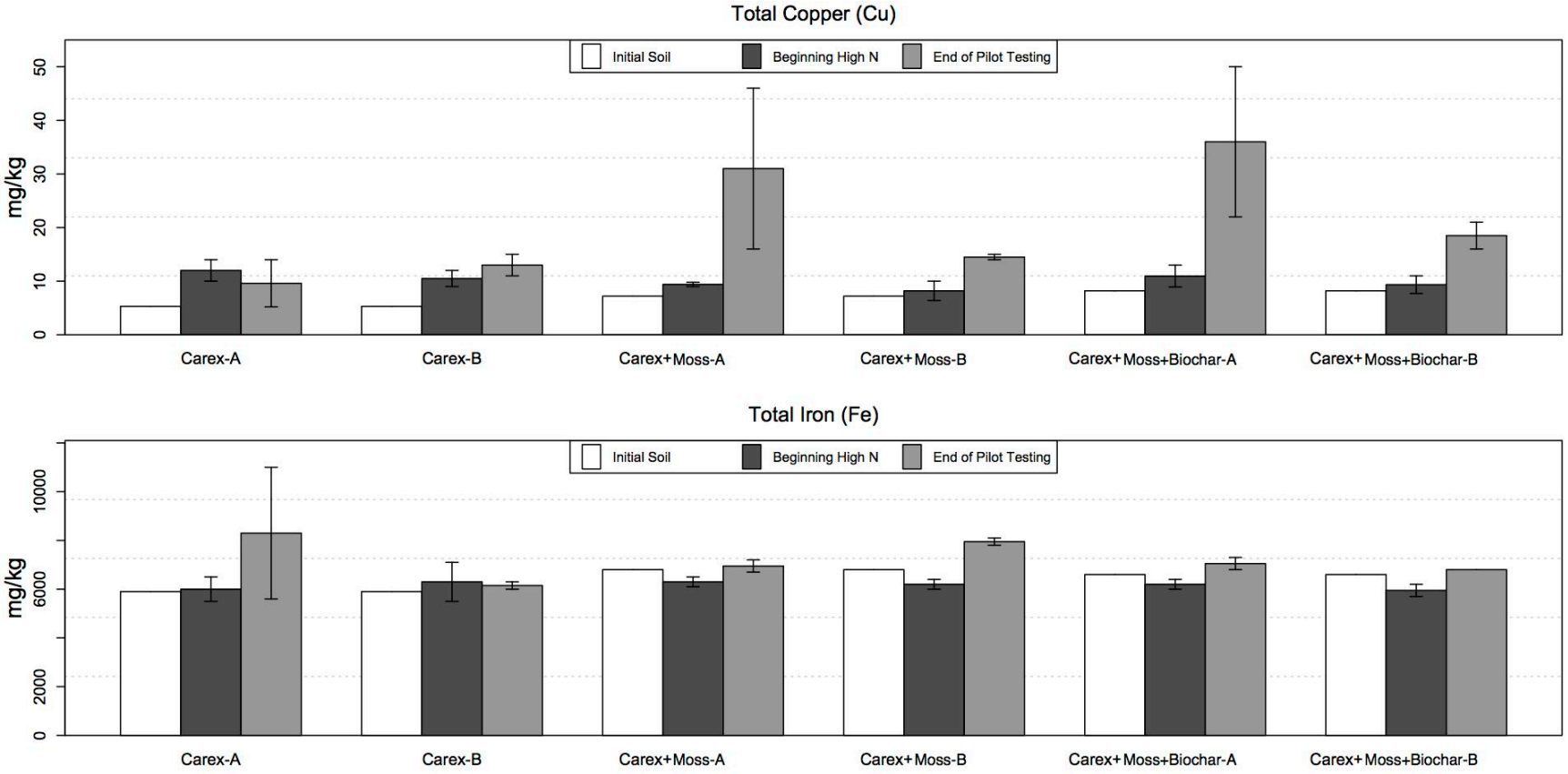


Figure 8: Copper, iron, selenium and sulphur concentrations in sediment throughout the pilot testing. Initial soil is the soil that was used to build the pilot systems, before any water was introduced. A and B cells for system designs are plotted separately, with error bars indicating the maximum and minimum values. The dotted red line indicates the selenium detection limit, as several samples were below this level. Selenium and sulphur plots are on the following page.



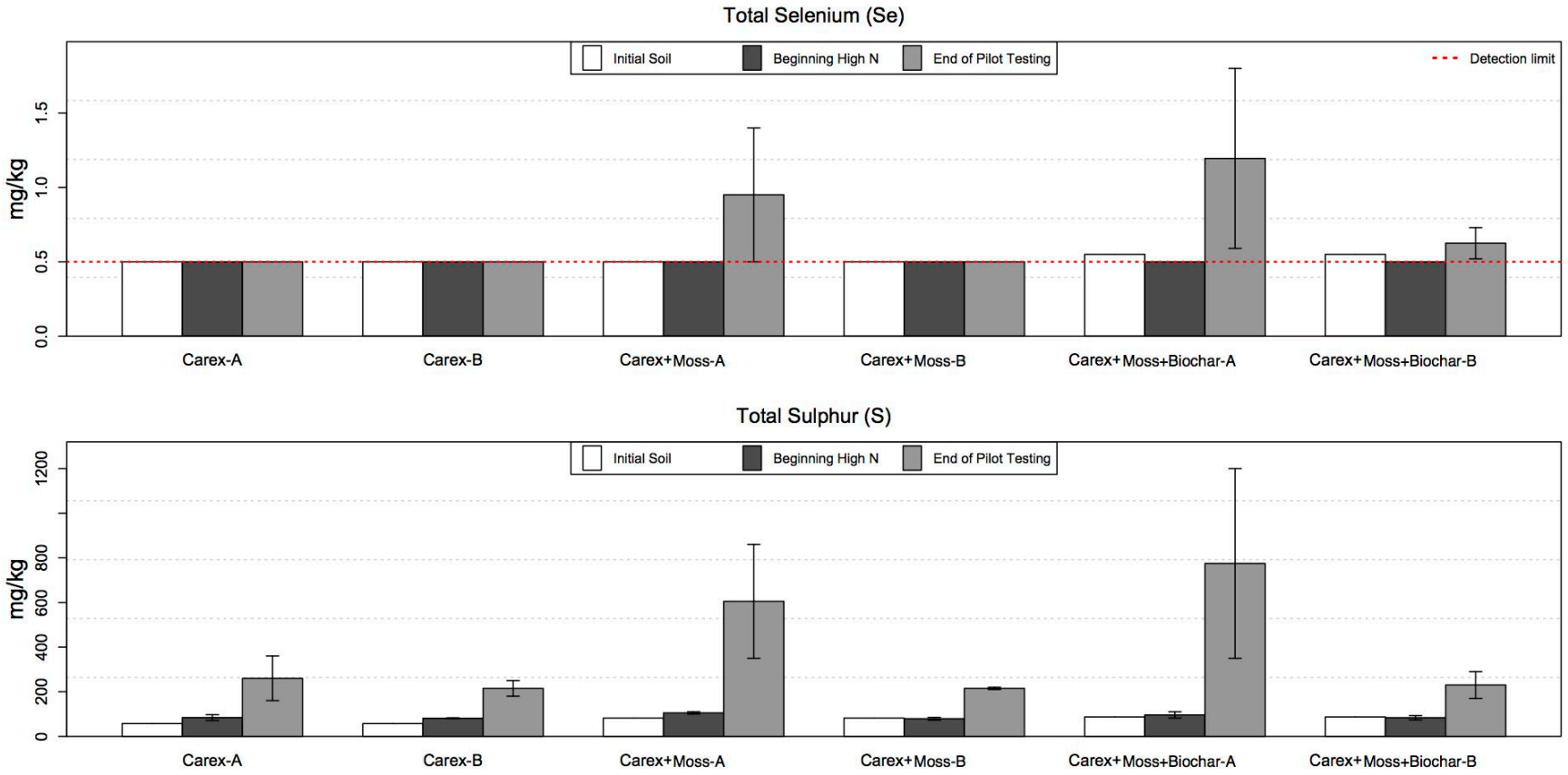

Total Selenium (Se)
Initial Soil
Beginning High N
End of Pilot Testing
Detection limit
mg/kg
0.0
0.5
1.0
1.5
Carex-A
Carex-B
Carex+Moss-A
Carex+Moss-B
Carex+Moss+Biochar-A
Carex+Moss+Biochar-B
Total Sulphur (S)
Initial Soil
Beginning High N
End of Pilot Testing
mg/kg
0
200
400
600
800
1200
Carex-A
Carex-B
Carex+Moss-A
Carex+Moss-B
Carex+Moss+Biochar-A
Carex+Moss+Biochar-B

Figure 9: Mass balance of copper in the pilot CWTS at the end of testing. The total amount of copper in the top 6 cm of sediment in A and B cells is shown for each system design, as well as the *Carex* in A and B cells. "Other" is the portion that was not analyzed for metals, and therefore cannot be differentiated. Examples of locations in the "Other" category would be plant roots, biofilms, and deeper sediment, as all of these were not analyzed. Additionally, it should be noted that the Moss did not survive the hybrid bioreactor-CWTS period, and therefore was not tested at the end of the pilot testing.

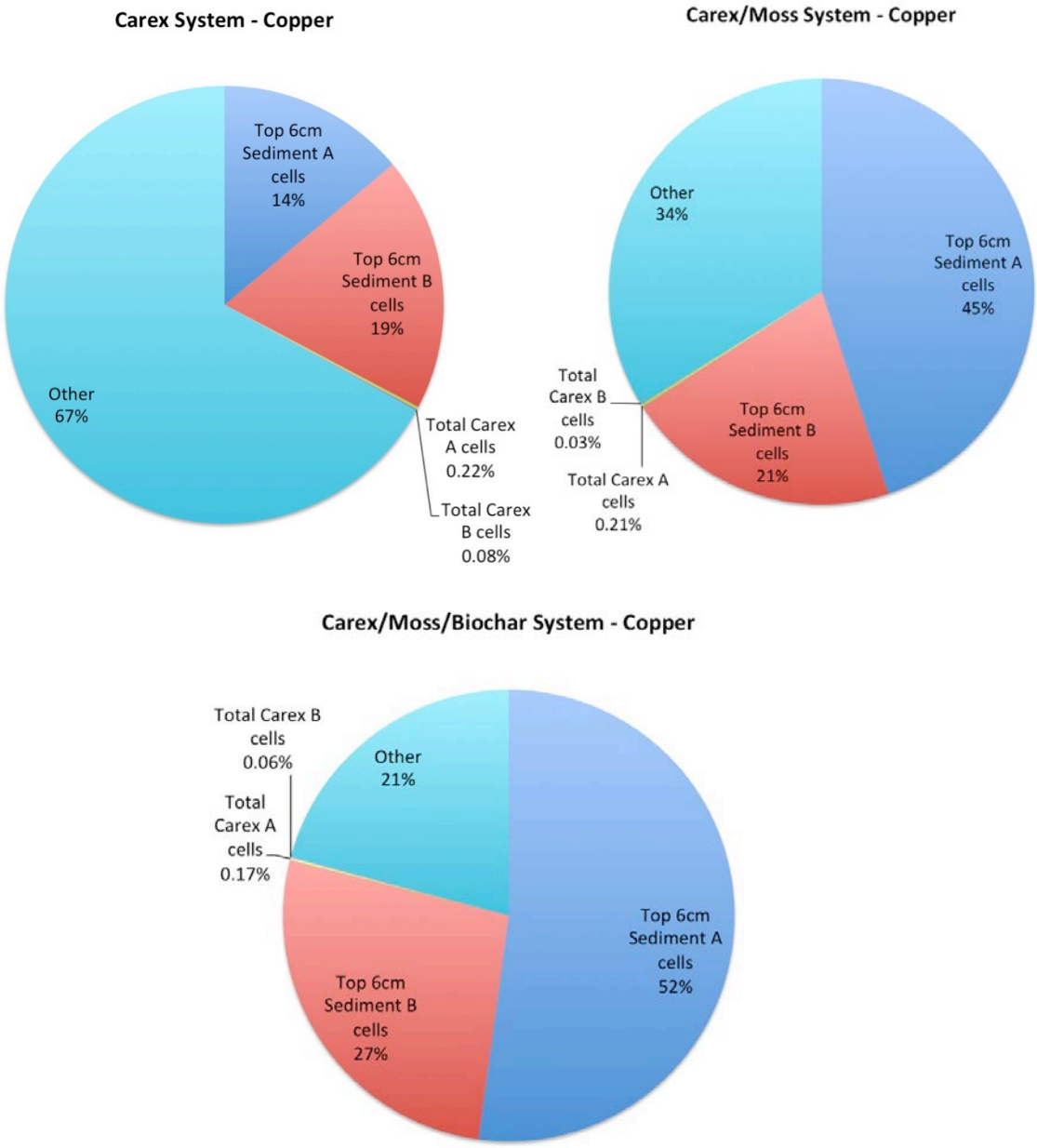


Figure 10: Mass balance of selenium in the pilot CWTS at the end of testing. The total amount of selenium in the top 6 cm of sediment in A and B cells is shown for each system design, as well as the *Carex* in A and B cells. "Other" is the portion that was not analyzed for metals, and therefore cannot be differentiated. Examples of locations in the "Other" category would be plant roots, biofilms, and deeper sediment, as all of these were not analyzed. Additionally, it should be noted that the Moss did not survive the hybrid bioreactor-CWTS period, and therefore was not tested at the end of the pilot testing.

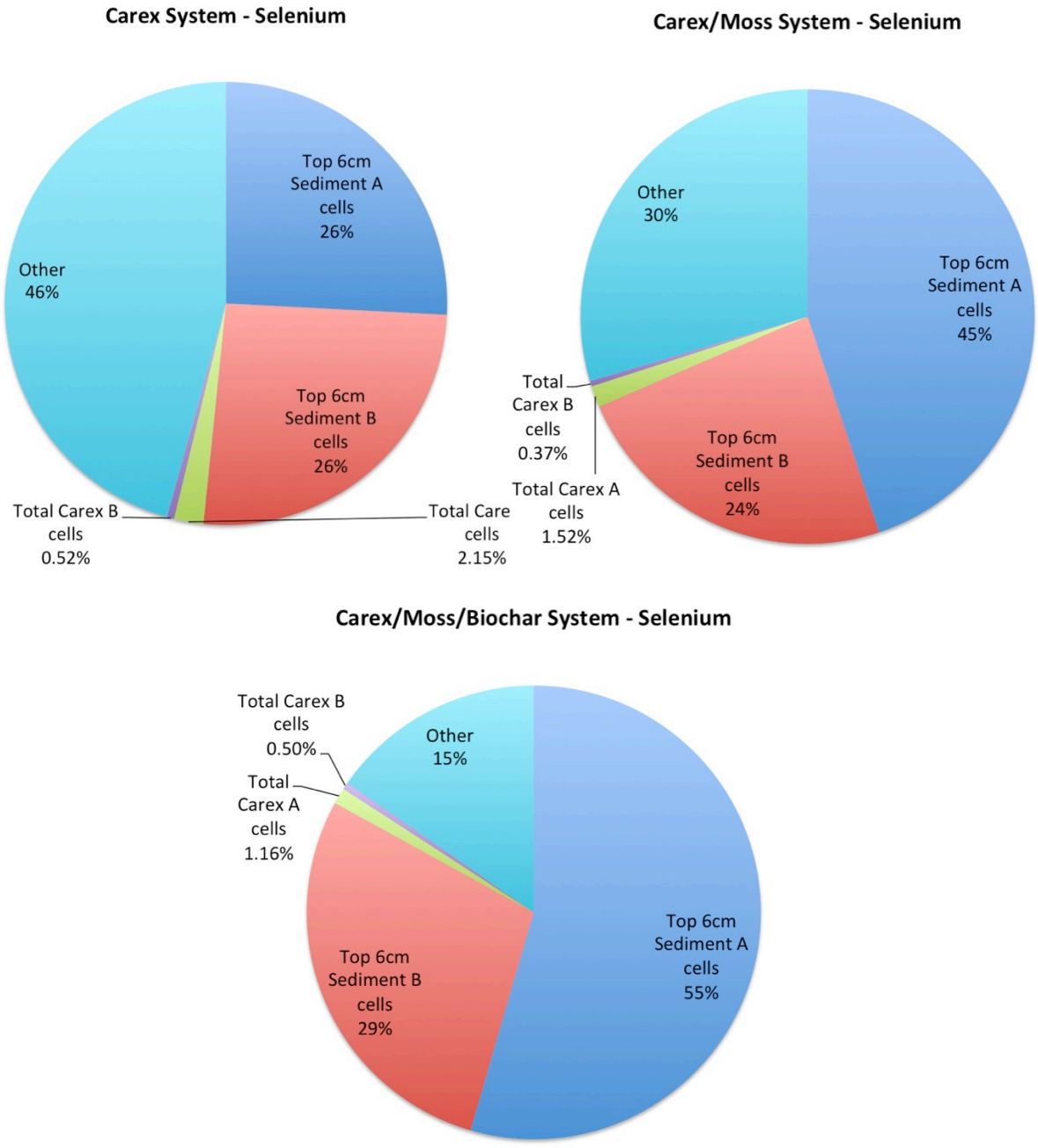


Figure 11: Comparison of bioconcentration factor for copper and selenium in *Carex* in A and B cells from each pilot system design. The p-value is calculated using Friedman test, with significant results shown in red. A significant test means that a difference between the BCF in A and B cells is found (e.g., A cells are always higher than B cells, or vice versa).

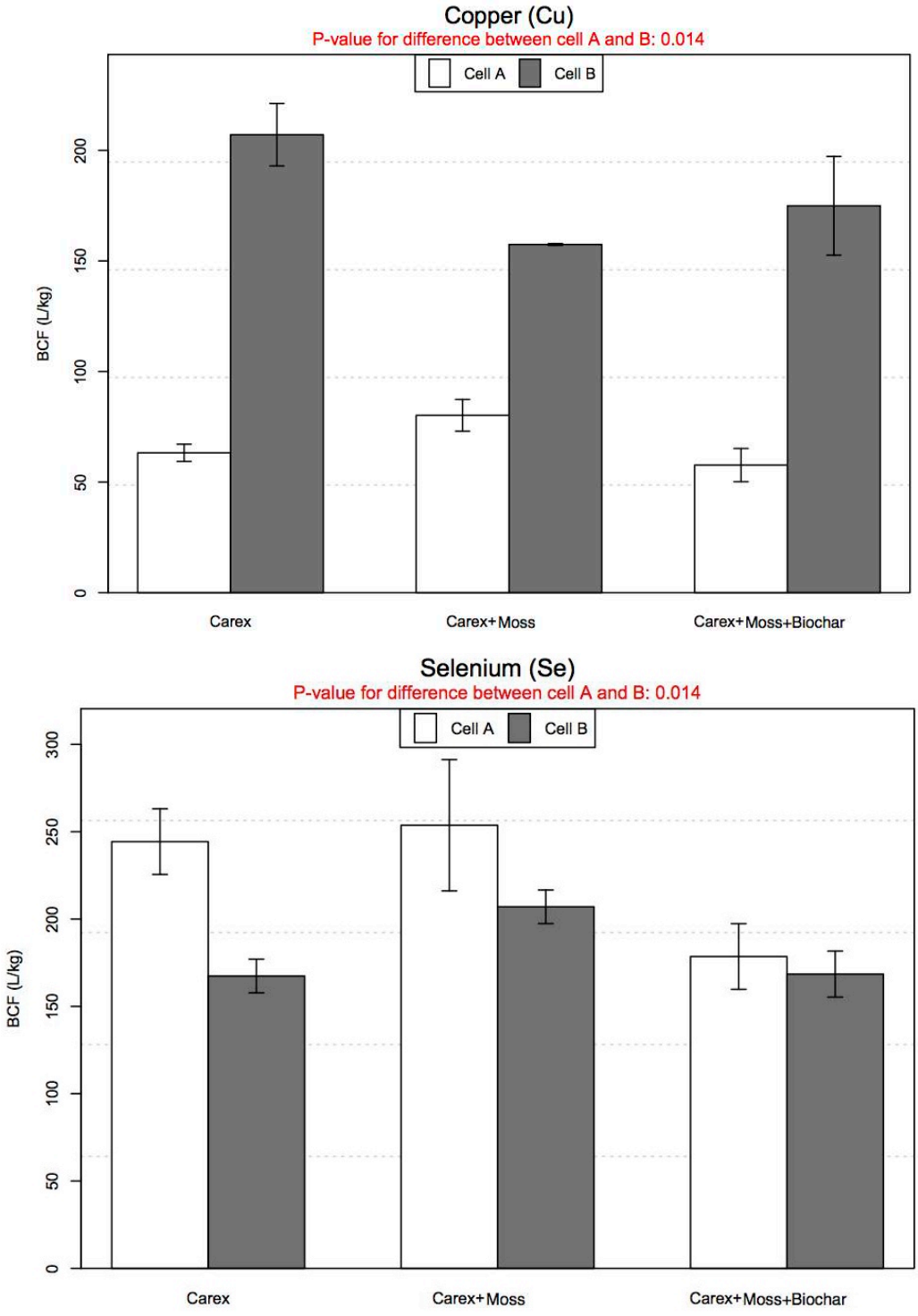


Figure 12: Copper and selenium concentrations in *Carex* before and after the freeze-thaw test, for all three system designs.

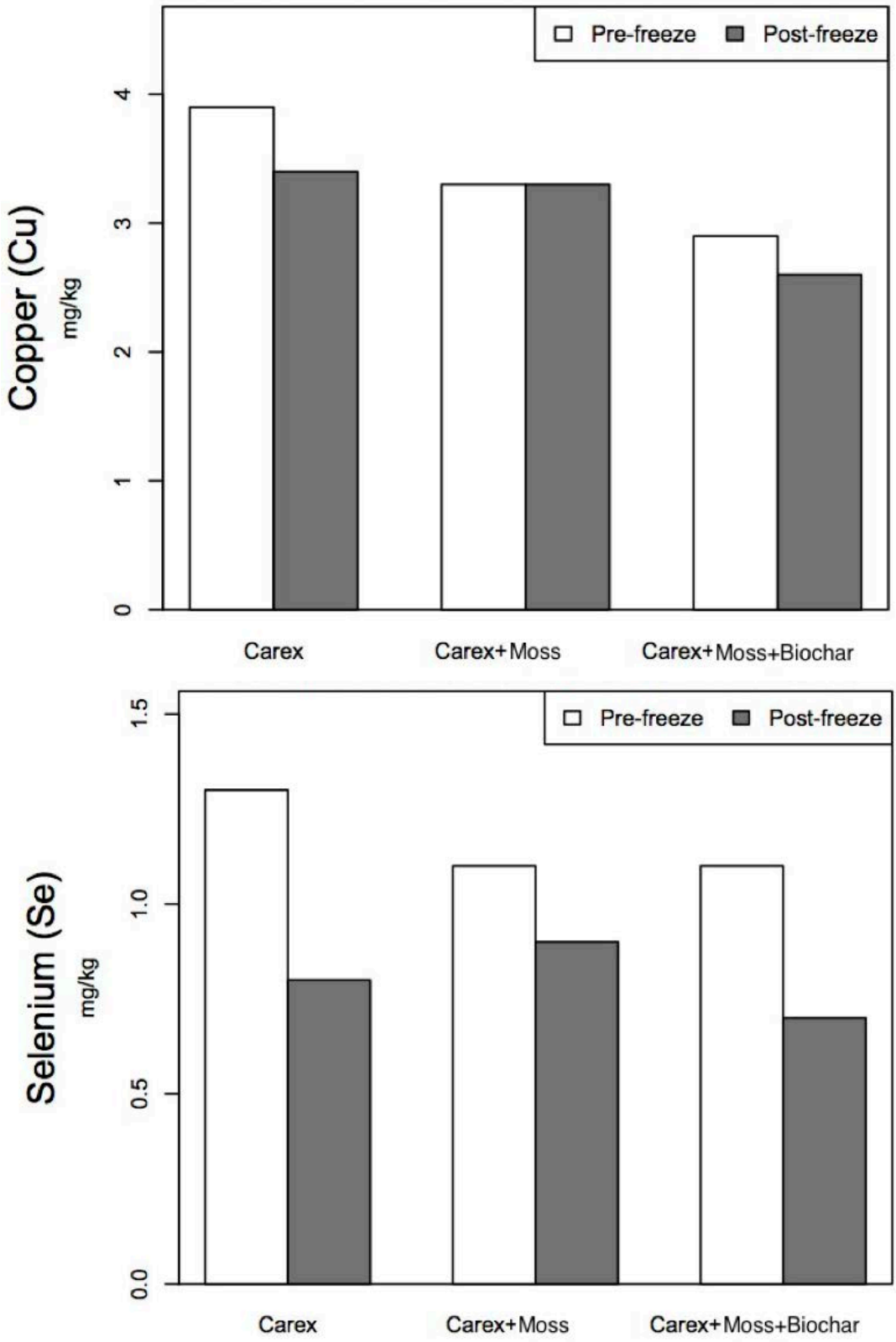


Figure 13: Copper and selenium concentrations in Moss before and after the freeze-thaw test, for the two system designs that had Moss.

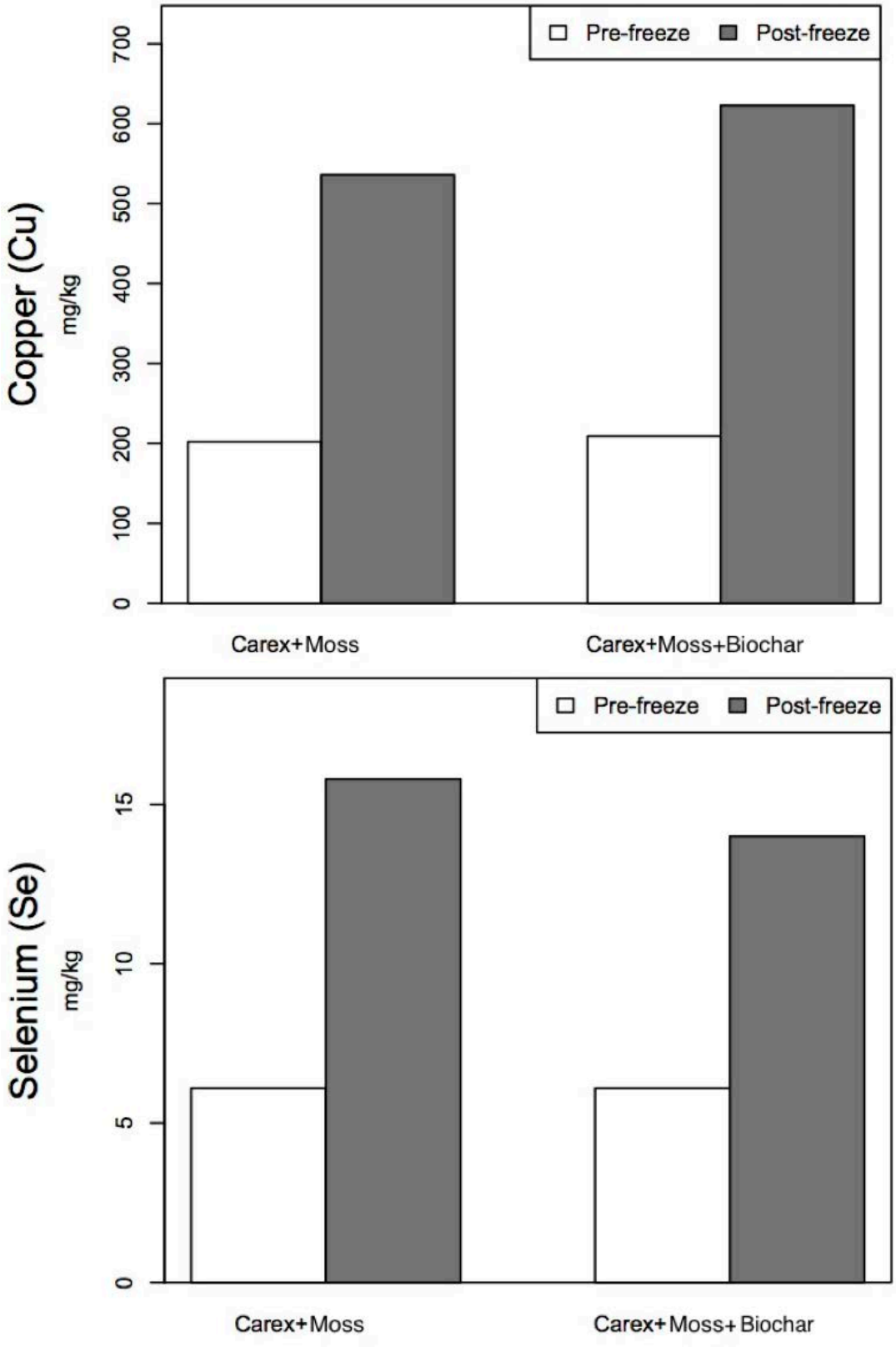


Figure 14: Photograph of *Carex aquatilis* (Sedge) roots at system takedown



Figure 15: Photograph of CWTS cell during takedown of the system. Orange-colored hydrosoil layer is approximately 18 cm, black detritus toproot zone is ~3 cm. Water depth was ~20 cm during the high and low nitrogen periods, and ~30 cm above the sediment during the hybrid bioreactor-CWTS period.



Figure 16: *Carex* stem counts (top) and height (bottom) of each system design, during all periods of testing.

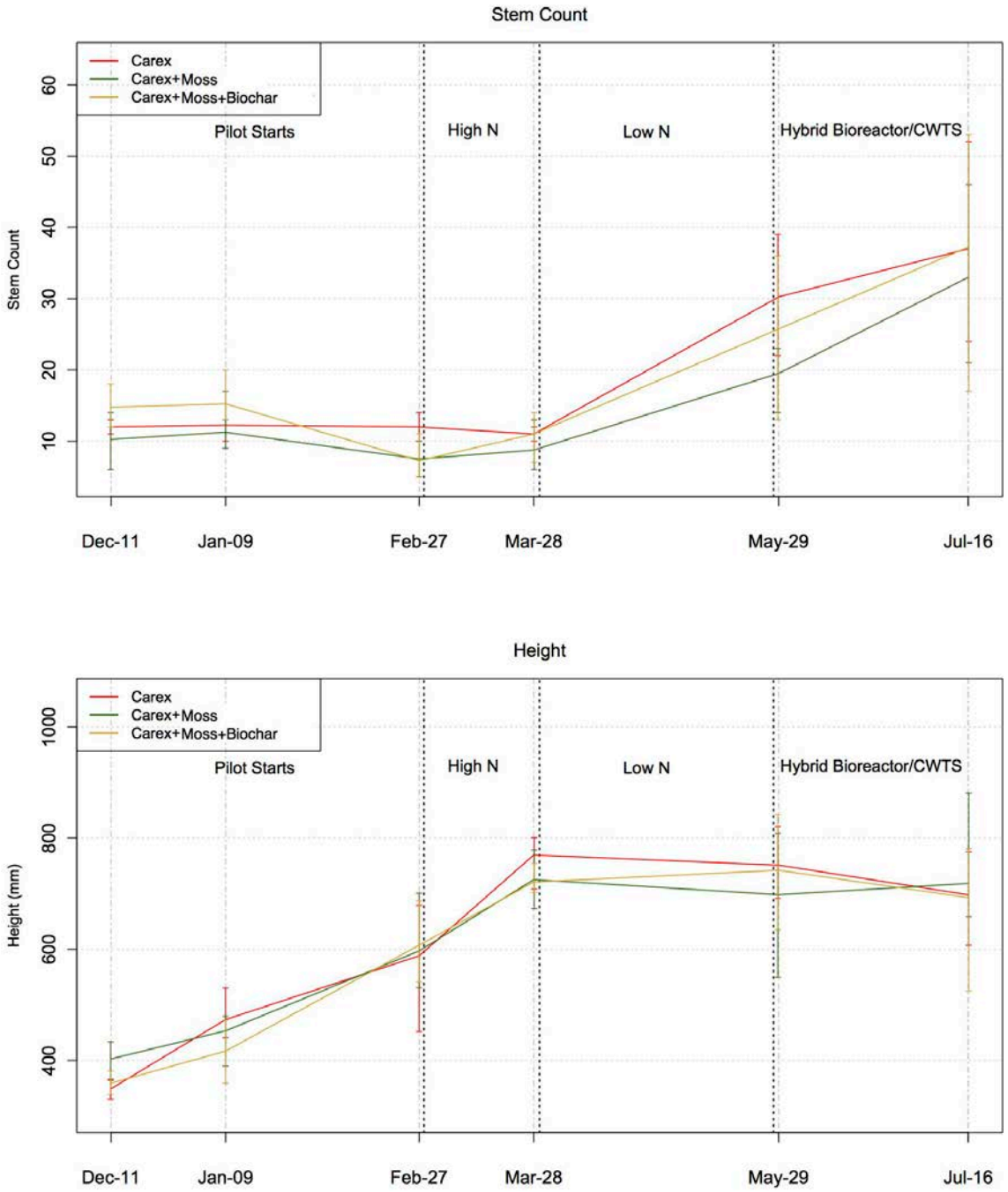
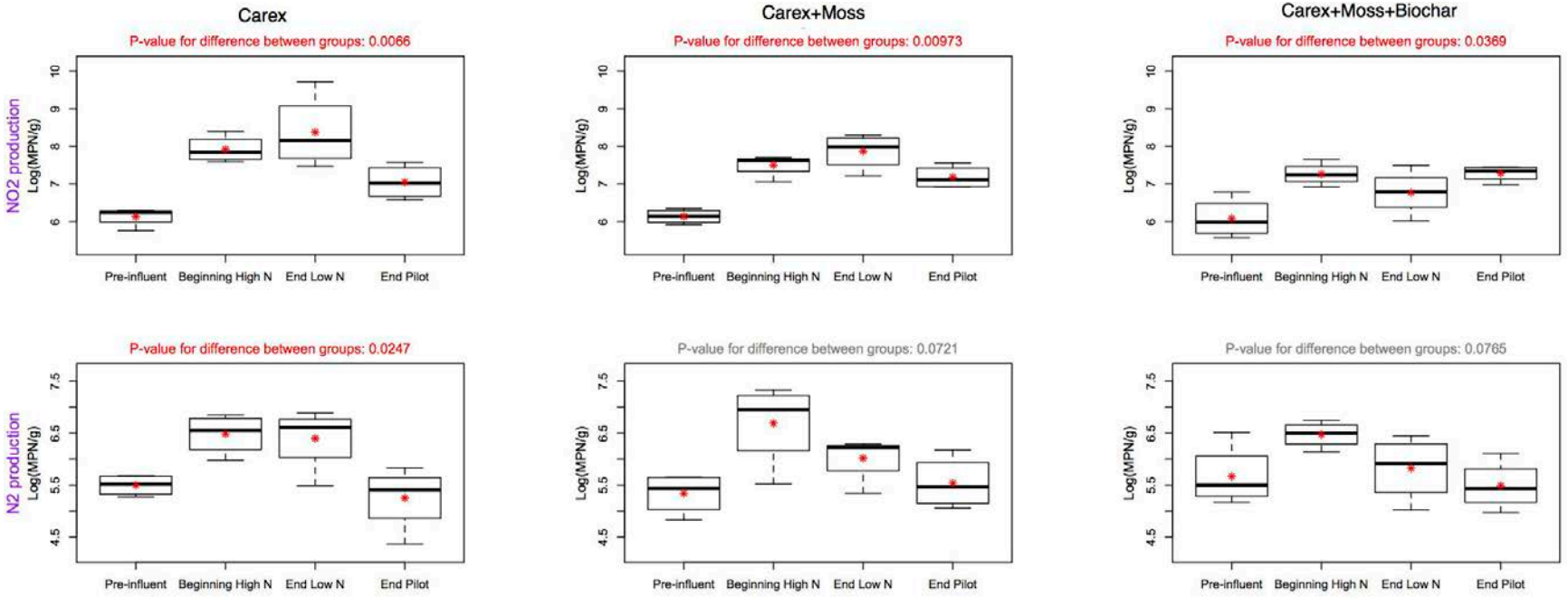
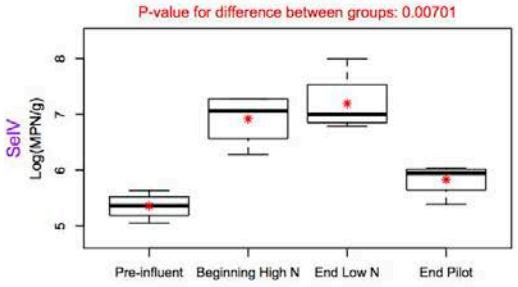
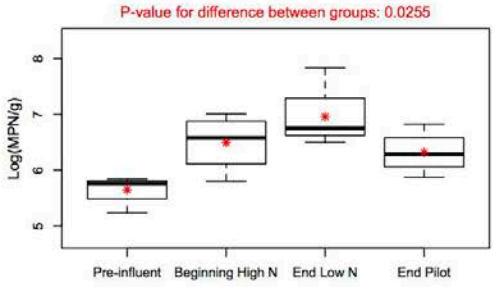
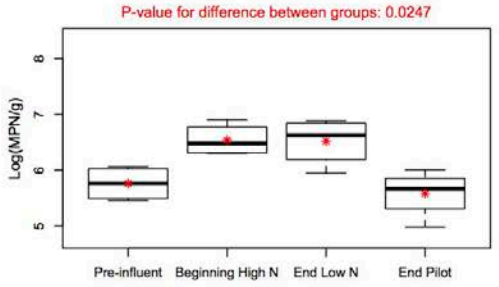


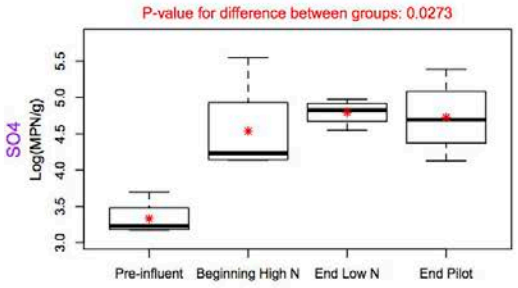
Figure 17: Comparison of the most probable number (MPN) of nitrate-, selenite-, and sulphate-reducing organisms in each system over time. NO_2 production indicates the MPN of organisms that are capable of reducing nitrate to nitrite. N_2 production shows the MPN of organisms that can reduce nitrate to nitrogen gas. The MPN was compared among the four time periods, and a P-value was calculated using the non-parametric Kruskal-Wallis test. The first plot represents systems with just *Carex*, the second plot is systems with *Carex* and Moss, while the third plot is systems with *Carex,* Moss, and biochar. Significant differences (P<0.05) are indicated in red, meaning a significant difference is found between the four testing periods in a given system design. Box plots are used to visualize the results; top and bottom edges of the box show the 25th and 75th percentile, respectively, while the thick black line depicts the median and the red asterisk shows the mean. The upper and lower whiskers provide the maximum and minimum non-outlier values. The remainder of the figure is provided on the following pages.

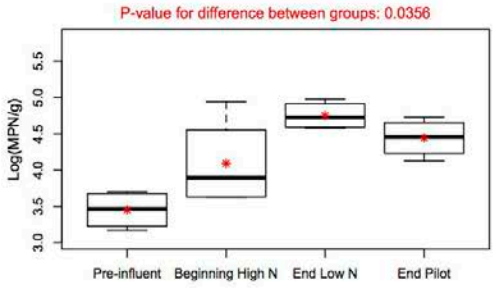


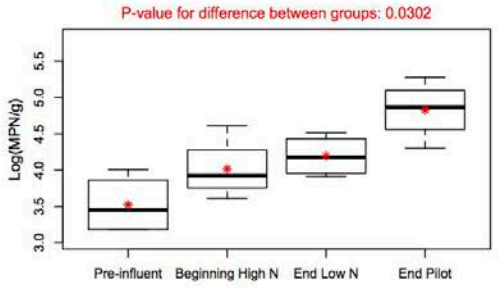

P-value for difference between groups: 0.00701
SeIV
Log(MPN/g)
5
6
7
8
Pre-influent
Beginning High N
End Low N
End Pilot


P-value for difference between groups: 0.0255
Log(MPN/g)
5
6
7
8
Pre-influent
Beginning High N
End Low N
End Pilot


P-value for difference between groups: 0.0247
Log(MPN/g)
5
6
7
8
Pre-influent
Beginning High N
End Low N
End Pilot


P-value for difference between groups: 0.0273
SO4
Log(MPN/g)
3.0
3.5
4.0
4.5
5.0
5.5
Pre-influent
Beginning High N
End Low N
End Pilot


P-value for difference between groups: 0.0356
Log(MPN/g)
3.0
3.5
4.0
4.5
5.0
5.5
Pre-influent
Beginning High N
End Low N
End Pilot


P-value for difference between groups: 0.0302
Log(MPN/g)
3.0
3.5
4.0
4.5
5.0
5.5
Pre-influent
Beginning High N
End Low N
End Pilot

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Minto Mine Constructed Wetland Treatment Research Program – Demonstration Scale

Document – 011_0315_01A

capstone MINTO MINE

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1. Introduction and Background

The Minto mine, operated by Capstone, is located 240 km northwest of Whitehorse on the west side of the Yukon River. The Minto property lies within the eastern part of the Dawson Range, with elevations from 700 to 1000 m; the landscape has rounded mountains intersected by broad valleys and drainages that are part of the Yukon River watershed.

The Minto mine has been in commercial operation since October 2007 and the deposits being mined are copper sulphide mineralized zones. Surface and groundwater water quality is a key consideration in the evaluation of potential effects of mining and mineral development projects and changes to water quality parameters have the potential to affect aquatic and human use of water resources. A Reclamation and Closure Plan (RCP) is required under both the Water Licence and the Quartz Mining Licence. The RCP is intended to address the long-term physical and chemical stability of the site and closure of the proposed features and disturbances associated with the mine. As a part of the RCP, a Constructed Wetland Treatment System (CWTS) is being designed, evaluated, and optimized for water treatment at closure through a phased program (Minto Phase V/VI Expansion Project, YOR Project Number 2013-0100).

Once established, wetlands can become self-sustaining ecosystems with plants providing yearly renewal of carbon to fuel microbial activity. As such, they possess the desirable potential to remediate contaminated mine drainage for as long as it is generated. In order for CWTSs to be effective, they must be designed, piloted, optimized, implemented, and maintained in a site-specific manner. A scaled approach for CWTS implementation allows for improvement, optimization, and flexibility for modifications along each step. Phases include: 1) site assessment and information gathering, 2) technology selection and conceptual design, 3) pilot-scale testing and optimization (controlled environment), 4) on-site demonstration-scale confirmation and optimization, and 5) full-scale implementation. Phases 1-3 have been completed (reports 2013-0100-256 and 2013-0100-257 on YESAB registry, and Contango, 2014) and confirmed plant amenability to transplantation and the CWTS design for further on-site testing. During pilot-scale trials, the selected CWTS design achieved 92% removal of copper (mean influent 146 µg/L, outflow 11.3 µg/L) and 41% removal of selenium (mean influent 10.2 µg/L, outflow 6 µg/L) using synthetic influent designed to mimic the worst-case water chemistry of a long-term closure scenario. Phase 4 of the project is now underway, with the on-site demonstration scale CWTS constructed at the Minto Mine during fall 2014. This document reports on the on-site demonstration scale CWTS design, construction, preliminary data, sampling schedule, and the long-term conceptual closure plan.

2. Design

2.1. System layout and dimensions

The demonstration-scale CWTS includes 2 systems in parallel with 2 cells in each series and a final catchment basin that both systems flow into (Figures 1-3). The location of the system in relation to the Minto site is provided in Figure 1. A stable foundation on waste

rock fill was selected for the CWTS and a base of residuum (sandy gravel) material was placed in compacted lifts to allow for shaping of the structure to design specifications. The two parallel systems serve as a replicate for data analysis, but as testing progresses, it is also possible that one of the two systems may be subjected to alternate conditions for comparative purposes.

The four planted cells ranged slightly in width and length. The as built sizing of the cells (at soil surface), after soil and amendments were added, ranged from 2.8 m to 3.8 m in width and 8.7 m to 9.8 m in length (Figure 4, Table 1). This was consistent with the submitted design of the cells being 3 to 4 times longer than wide.

The 4 cells and catchment basin were roughed in and large sharp rocks hand picked and removed prior to final grading and survey. The finished surface was lined with an impermeable 4030 Enviro Liner® that was welded together and sandwiched between two layers of 12 ounce geotextile fabric.

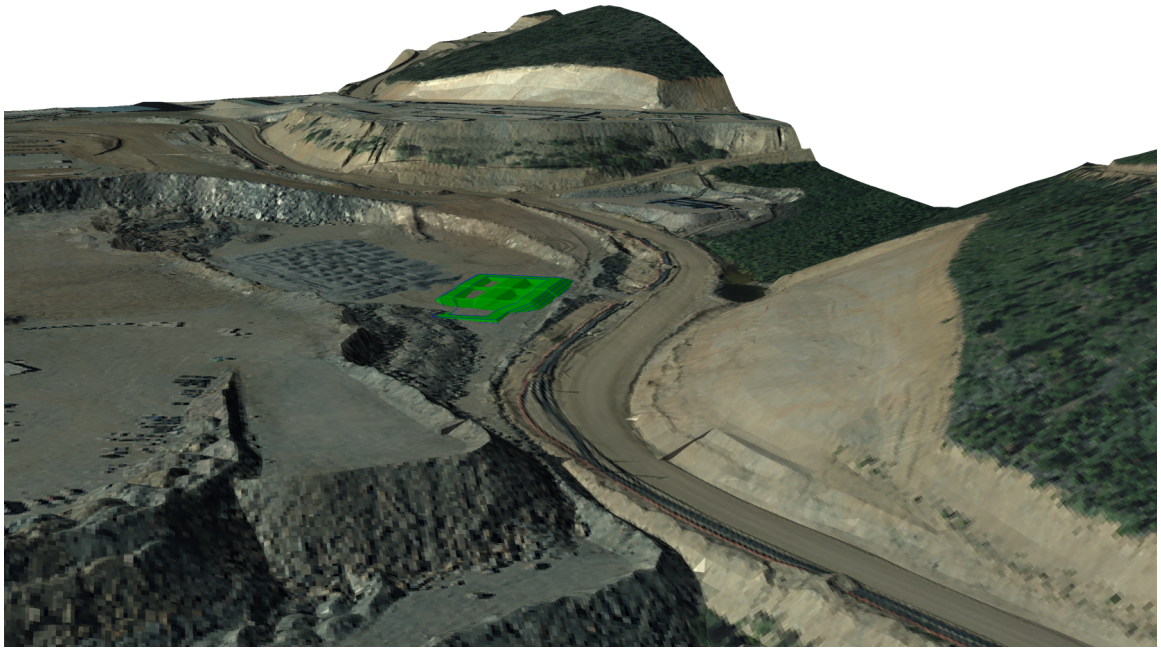


Figure 1. Location of demonstration-scale CWTS at Minto mine.

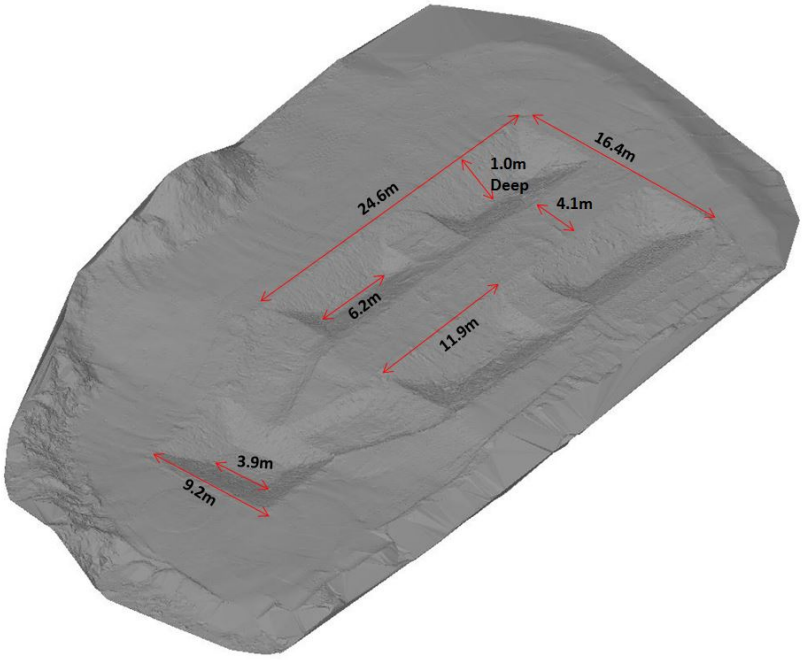


Figure 2. As built schematic of demonstration-scale CWTS.



Figure 3. Demonstration-scale CWTS prior to liner installation.

Table 1. Measurements of demonstration-scale CWTS cells at soil surface and resultant areas of treatment systems.

Measurement		1A	1B	2A	2B
Width (m)	Inflow	3.6	3.8	3.6	2.8
	Outflow	3.3	3.8	3.3	3.6
Length (m)		9.8	9.5	8.7	8.8
Approximate surface area at soil (m^2)		33.8	36.1	30.0	28.2
Total area of System at soil (m^2)		69.9		58.2	

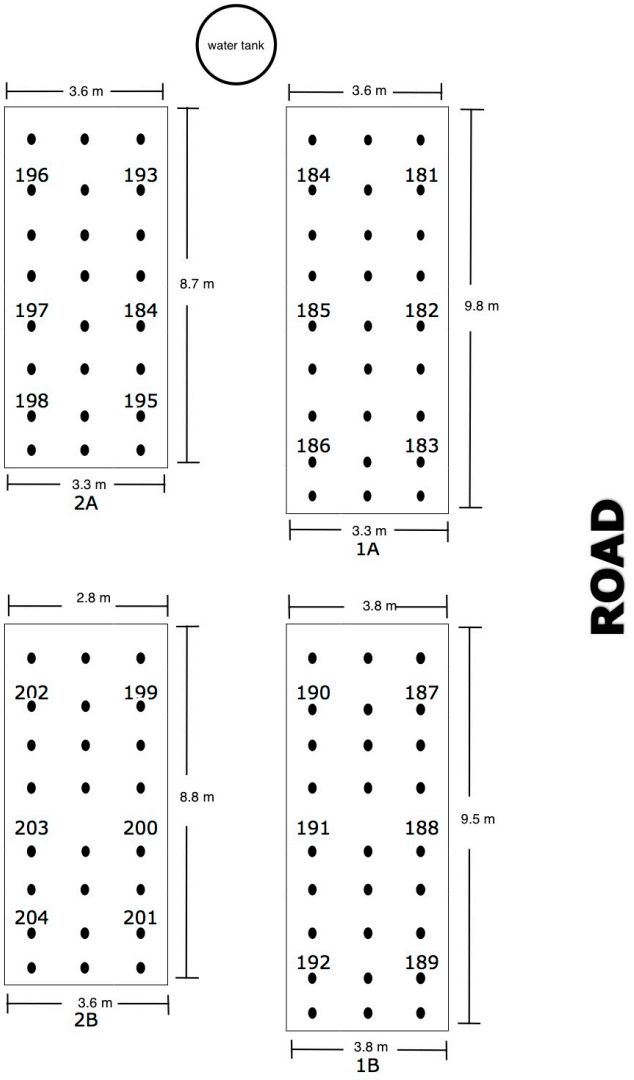


Figure 4. Diagram of demonstration-scale CWTS with dimension measurements at soil surface showing grid (by moss stakes; black dots) and locations of soil redox probes (with identifying numbers; Table 2).

2.2.Soil and Amendments

The recommended soil for the CWTS is sand, with 2-7% by volume as organic material (e.g., woodchips, peat). In the pilot-scale systems, this resulted in a total organic carbon (TOC) content of 0.2-0.6% (the sand itself was at 0.1% TOC prior to adding amendment). In this composition, the sand allows for circulation of water through the root zone of the plants, while the organic matter initiates reductive microbial processes that in future years will be sustained by decaying plant matter. In the case of this demonstration-scale system, the soil added to each of the 4 cells was from a local overburden site. The characteristics of this soil are provided in Appendix A. This borrow site was different than the site that was sampled during the Contango Strategies Ltd. (CSL) assessment in 2013, and also different than that tested in June 2014 (Appendix A, Table A1-A3).

The overburden soil used in the demonstration-scale CWTS was taken from stockpiled material that originated as richly organic soil from stripping of the Area 2 Pit. The calcium, magnesium and sulphate concentrations were higher than those of the pilot-scale tests, but this is not expected to negatively impact the performance of the CWTS. Owing to the general mineralization of the area, the concentration of copper was elevated in all soils tested as potential borrow sources. The synthetic precipitation leaching procedure (SPLP) analysis conducted on the soils used in the demonstration-scale wetland suggested there is between 148-608 µg/L of available copper and 0.35-0.69 µg/L of available selenium in the soil. This should be monitored and taken into consideration with wetland performance, and it may be necessary to investigate other sources of soil with lower copper concentration for final full-scale construction.

It was observed that the soil that was used for the demonstration-scale wetlands was higher in organic content and lower in sand than those used in the pilot-scale testing, or tested during the site assessment or June 2014 pre-construction sampling. This higher organic content was also associated with a more loamy texture. Due to the higher organics content in the soil being used for the demonstration-scale CWTS, there was less organic amendment added (by volume) than was initially planned. As the soil was added by the backhoe (Figure 5), there was 1 standard size straw bale (measuring 36 cm x 46 cm x 102 cm) and 50 rounded standard shovels of a 50:50 mix of spruce and pine wood shavings placed in each of the cells and mixed by hand with shovels and rakes (Figures 6 and 7). There was no biochar added as was proposed in the initial design as this was determined to not be required after the piloting phase at the CSL facilities (Contango, 2014).

The high organic content of this soil may actually be beneficial to the treatment of elements, through both sorption and increased microbial reductive processes. The drawback of using this organic material is that it may not facilitate the circulation of water in the root zone the way sand or gravel would, and it is also more difficult to work with for construction purposes. Specifically, the material used in the demonstration-scale CWTS did not support weight as a sandy soil would. For this reason, the wetland was planted while the soils were somewhat dry (before flooding), as the people planting the wetland would sink into the soil if it were wet. This should be taken into consideration if building the full-scale CWTS for the sake of ease of accessing monitoring points within the wetland, or if any revegetation

campaigns are needed. Over time, however, the plant roots and moss will grow and provide greater physical stability to the soils.

Upon inspection, there was too much soil placed in the cells by the mini-backhoe; some of the soil was therefore removed to bring the outflow depth to 20 cm and maintain a level grade in each of the 4 cells to a tolerance of +/- 5 cm in order to provide a uniform water depth. A uniform water depth is necessary to prevent future issues such as channeling of water, open water spots due to plants being unable to grow because of excessive water depth, or conversely, drying out.



Figure 5. Soil being added to demonstration-scale CWTS cells.



Figure 6. Organic soil amendments used in demonstration-scale CWTS. Left, wood shavings 50:50 pine and spruce mixture; Right, straw.



Figure 7. Mixing organic amendments into soil in demonstration-scale CWTS cell 2A.

2.3.Plants

The demonstration-scale CWTS was planted with *Carex aquatilis* (Sedge) and mosses, as was determined by pilot-scale testing as being the best design for this site and application (Contango, 2014). These plant species were harvested for the demonstration scale CWTS from a location that was identified during the August/September 2013 site assessments in the vicinity of the W10 monitoring point (reports 2013-0100-256 and 2013-0100-257 on the YESAB registry). This site had *C. aquatilis* and aquatic mosses growing together as is

desired for the CWTS design (Figures 8 and 9). Plant tissues were sent for analytical testing to determine initial concentrations (Appendix A, Table A4).

All 4 cells of the demonstration-scale CWTS were measured and the center of the width marked at both the inflow and outflow ends of each of the cells. The tape measure was then placed on the center axis down the length of each cell to start laying the center grid. The grid was created using wooden gardening stakes (50 cm in height), with aquatic moss tied to one end and inserted into the soil so that the moss was just above the soil (Figure 10; eventually to be submerged in the water). The grid for planting the *C. aquatilis* was created with these stakes, starting on the inflow end of each cell, with stake placement every 1 m progressing to the outflow (Figure 11). The final grid near the outflow of the cell was therefore sometimes less than 1 m of soil, but may look like a greater size due to the overlaying water. This was repeated moving outward from the center line, creating a grid of 1 m square areas for planting *C. aquatilis* to facilitate future monitoring of plant density. Depending on the varying width and length of the cells, along the sides of the length of the cells and the outflow of the cells the grids varied from 1 m squared to smaller than 1 m squared (Table 2).

Table 2. Number of planted grids in each demonstration-scale CWTS cell.

Item	Cell			
	1A	1B	2A	2B
Moss stakes	3*9=27	3*9=27	3*8=24	3*8=24
1 m squares	18	18	16	16
< 1 m squares (perimeter of CWTS cells)	22	22	20	20

Using the 4 stakes marking the edges of the grid, 5 *C. aquatilis* were planted into the soil inside each 1 m by 1 m grids, with one in the middle and 4 others approximately equidistant from the stakes to create an even distribution of 5 sedge per square meter (Figure 12). This placement and grid system was used to ensure even distribution of plants through the demonstration-scale CWTS, with the foresight of future monitoring for plant establishment and density. As the grids at the perimeter of each wetland cell varied and were often smaller than 1 m by 1 m, it was estimated based on the size of the grid how many sedge should be planted to maintain a consistent density of plants.



Figure 8. Harvesting plants from W10 location (top), and transporting back to demonstration-scale CWTS (bottom).



Figure 9. Images of *Carex aquatilis* and aquatic mosses harvested at W10 for planting the demonstration-scale CWTS. Left, aquatic mosses growing at water surface at edge of *C. aquatilis*; top right, underwater photograph of aquatic moss; bottom right, *C. aquatilis* and aquatic mosses growing together.



Figure 10. Example of moss tide to stakes, which then were used to make grid in each demonstration-scale CWTS (Figure 4, 11, 12 and Table2).



Figure 11. Moss stakes placed in demonstration-scale CWTS cell 2A, looking towards cell 2B.



Figure 12. *C. aquatilis* planted within grid of moss stakes in demonstration-scale CWTS cell 2A, and influent perforated pipe being laid in place.

2.4.Water

Water from the W36 area receiving seepage from the toe of the MVFE (Figure 13) was selected for the demonstration-scale CWTS testing as the leachate is similar to that expected upon closure in the Mill Valley Fill Extension (MVFE) area. The chemistry of this water at the time of bringing the demonstration-scale CWTS online (September 18, 2014) is provided in Appendix A, Table A5. A Grundfos CR1S-6 stage pump was selected to maintain the appropriate head of water in the holding tank to provide a range of flows between 0.1-1.1 m^3/hr on a 4-inch line with 30 m of head (Figure 13). The 3,000 Liter holding tank sits above and between the inflow of cells 1A and 2A and is kept at a constant head via pumping from the W36 sump (Figure 14). Overflow water is returned by gravity flow in a pipe to the W36 area. Meanwhile, water within the holding tank is gravity fed through control valves and flow meters prior to entering the parallel treatment cells (Figure 14). The tank was initially rinsed twice with the W36 water before releasing water into the demonstration CWTS because it had originally contained road dust suppressant (Envirobind). Water is constantly pumped from W36 to the tank that has been modified to maintain a constant head (and therefore pressure) to keep flow rate consistent when set using the manual valves. Flow rates are measured using ultra low flow and high accuracy/totalizer meters from GPI (Figure 14).

A flexible 1-inch braided rubber hose with perforations was attached to each of the flow meters and run to the inflow of cells 1A and 2A (Figure 12 and 14). The placement of these perforated hoses was adjusted to promote even flow across the width of the A cells. The system then operates by gravity flow, with Cell 1A and 2A flowing into 1B and 2B, respectively, before flowing into the catchment basin and returning by pipe to W36 (Figure 15).

Sandbags were used at the outflow of each cell to correct for unevenness in grade and prevent channeling to the lowest point. The average height of water in the cells above the soil was approximately 20 cm (Figure 16).

The flow rates for the systems were set to have a 10 day nominal hydraulic retention time (HRT), meaning water entering the wetland takes 10 days to exit. This is referred to as a nominal HRT because it is a calculation based on the size of the wetland and the amount of water entering, and not confirmed empirically using tracing dyes. The HRT was calculated based on the size of the CWTS at the soil, with a 20 cm overlay of water, but not adjusting for pore water because without dense vegetation, the pore spaces in the soil will only have minimal hydraulic relationship to the overlaying water. This HRT is much longer than what is necessary to achieve treatment based on the pilot-scale systems (~3 days), however, this was chosen to facilitate plant establishment and maturation. Because the sizes of the CWTS systems were slightly different, Series 1 (closer to the road) and Series 2 (further from road) were set with inflow rates of 13,928 L/day and 11,635L/day, respectively, both corresponding to an approximate 10 day HRT. As the flow meters record in US gal/min, this corresponds to 2.6 and 2.13 US gal/min.



Figure 13. Water source for demonstration-scale CWTS. Left, Grundfos CR1S-6 stage pump; Right, water source at W36.



Figure 14. Clockwise from left: holding tank, splitting control valve, and flow meters.



Figure 15. Demonstration-scale CWTS immediately after planting and filling with water (August 29, 2014). Top, photograph taken from camp behind kitchen looking down at inflow of CWTS; bottom, photograph taken from outflow holding pond looking towards camp in background.

2.5.Data Collection

Explanatory parameters are quantifiable aspects of a CWTS that can be used to assess the feasibility of treatment for a range of constituents, and therefore 'explain' the performance of a CWTS. These parameters, which often include acidity, alkalinity, conductivity, dissolved oxygen (DO), pH, oxidation reduction potential (ORP), ion balance, available electrons donors (e.g., organic carbon, reduced elements), and temperature, can be used to predict, promote, and/or optimize the ability of the system to treat different constituents.

The relative oxidation-reduction (redox) potential of the soil will be measured using inert electrodes (copper wire probes with platinum tips) permanently installed in the soil of the cells for the duration of the project (Faulkner et al., 1989; Figure 16). To take a reading of the relative redox potential between the soil and water, a reference electrode (Accumet Calomel) is suspended in the water column above the inert electrode and measured in millivolts by a voltmeter. There were 6 platinum tip probes distributed throughout each of the 4 demonstration-scale cells to try and achieve even distribution based on the varying cell sizes. The probes were inserted to cover the epoxy on the end of the probe and secured with flagging tape to the bamboo poles for visibility. The locations of each probe are provided in Figure 4 and Table 3.



Figure 16. Series 2 (cell 2B in foreground, cell 2A in background), showing sandbags used at inflows and outflows between cells.

Table 3. List of location and probe number for inert electrodes permanently installed in the soil of cells. General location also provided in Figure 4.

Stake - Row	Cell 1A	Cell 1B	Stake - Row	Cell 2A	Cell 2B
2nd - Inside	#181	#187	2nd - Inside	#193	#199
5th - Inside	#182	#188	5th - Inside	#194	#200
8th - Inside	#183	#189	7th - Inside	#195	#201
2nd - Outside	#184	#190	2nd - Outside	#196	#202
5th - Outside	#185	#191	5th - Outside	#197	#203
8th - Outside	#186	#192	7th - Outside	#198	#204

Probe numbers 205 to 209 were left as spares, in case of probe failure.

3. Sampling Schedule for 2015

The sampling schedule for 2015 was conceptually developed prior to beginning construction of the demonstration-scale CWTS (Table 4) and the analytical testing refined and summarized in Table 5. Actual dates of sampling will depend on timing of spring thaw and ability to bring the pumps online at the W36 pond to supply water to the demonstration-scale CWTS.

Table 4. Conceptual sampling parameters, locations, and frequencies.

Frequency	Parameter	Location	Sample Type
Weekly	Temperature (by data logger)	All Cells + Inflow to CWTS	Water
	pH		
	Dissolved Oxygen		
	Conductivity		
	ORP		
	Inflow rates/outflow rates (by meter)		
	Regulated Metals (ICP) (Total and dissolved)		
	Relative redox potential	All Cells	Soil
Monthly	Alkalinity	Outflow All cells + Inflow to CWTS	Water
	Hardness		
	Sulfate		
	Chemical Oxygen Demand		
	Total Organic Carbon		
	Ammonia		
	Nitrate/Nitrite		
	Total Kjeldahl Nitrogen		
	Biological Oxygen Demand	Outflow + Inflow of CWTS	
	Total Suspended Solids		

	Stem counts (and height)	All cells	Plant
Seasonally	Available NPKS	All cells	Soil
	Regulated Metals (ICP)		
	Total Organic Carbon		
	Cation Exchange Capacity (CEC)		
	Sodium Adsorption Ratio		
	Conductivity		
	Sequential Leaching		
	MPN for SeIV, NO_3, and sulphate reducing microbes, and total heterotrophs		
	Genetic microbial community profiles		
Twice per year	Detritus depth/accretion measurement	All Cells	Soil
	Regulated Metals (ICP)		Plant
	Available NPK and Sulphur		

Table 5. Summary of analytical sampling types, frequencies and locations

Water samples	
routine package: Ca, Mg, Na, K, Cl, SO_4, NO_3, NO_2, hardness, alkalinity, pH, EC, TDS	Monthly, outflow of each cell and inflow water
Chemical oxygen demand (COD)	Monthly, outflow of each cell and inflow water
Total organic carbon	Monthly, outflow of each cell and inflow water
Ammonia	Monthly, outflow of each cell and inflow water
Total Kjeldahl Nitrogen (TKN)	Monthly, outflow of each cell and inflow water
Total suspended solids	Monthly, outflow of series, and inflow water
Biological Oxygen Demand (BOD)	Monthly, outflow of series, and inflow water
Regulated metals water package (dissolved and total)	Weekly, outflow of each cell and inflow water
In-situ water testing (pH, DO, ORP, Conductivity)	Weekly, all cells and inflow water
Hydrosoil samples	
Relative soil redox	Weekly, all probes (6 per cell)
Cation exchange capacity (CEC)	Seasonally (3x per year), 3 places per cell
SAR, pH, EC, %sat, Ca, Mg, Na, K, Cl, SO4	Seasonally (3x per year), 3 places per cell
Available NPK and sulphur	Seasonally (3x per year), 3 places per cell
Total organic carbon	Seasonally (3x per year), 3 places per cell
Sequential Leaching	Seasonally (3x per year), 3 places per cell
Plant tissue samples	
ICP	2 plant types in triplicate (pre-planting in 2014)
Available NPK and sulphur	2 plant types in triplicate (pre-planting in 2014)
TOC	2 plant types in triplicate (pre-planting in 2014)

4. Long-term conceptual testing plan

A conceptual long-term testing plan has been developed for the demonstration-scale CWTS

(Table 6), which will be refined and adapted based on performance and scientific findings as the trials are conducted. The demonstration-scale wetlands will run for a minimum of 2 more years, until the end of 2016, but ideally longer in order to assess performance under a wider range of conditions. The conditions that could eventually be tested include both natural/environmental and selected influenced pressures, and can be imposed on the systems to mimic peak flow rates or droughts. In 2015 the systems will be allowed to continue to mature, with plants becoming more established and abundant, and microbial communities accordingly acclimating to the targeted conditions. Once plants are well established and showing signs of colonizing through the wetland, the water depth may be increased to 30 cm and the flow rate may also be increased. First, a 5 day HRT (i.e., twice as fast as currently) may be tested, and then fluctuations will be tested based on anticipated seasonal variation.

Table 6. Schedule as per proposed scope of work:

Item		Proposed	Actual
Identify potential location for demonstration scale CWTS (CSL site visit – 1 scientist)		June 1-14 2014	Completed
Engineering and geotechnical (Minto)		June - July 2014	Completed
Construction (Minto)		July 2014	Completed
Planting and bringing system online (CSL site visit – 1 scientist, 1 technologist), coordinate for local students to assist.		August 2014	Completed (no students available, brought 2 technologists)
Monitoring	2014	Acclimation and maturation at constant flow rate of 10 d HRT	Completed
		September - CSL site visit/checkup (1 technologist, 1 scientist)	Did not occur because constructed was last week of August.
	2015	Continued maturation/acclimation. Operation at constant flow rate of 5-10 d HRT	On Schedule
		Spring – CSL site visit/checkup (1 technologist, 1 scientist), includes micro sampling	
		Summer - Increase depth from 10 cm to 20 cm (1 technologist), includes micro sampling	
		Fall – CSL site visit/checkup (1 technologist), includes micro	

		sampling	
	2016	Operation with weekly flow variations based on seasonal variations	
		Spring – CSL site visit/checkup (1 technologist), includes micro sampling	
		Summer - CSL site visit/checkup (1 technologist, 1 scientist), switch to hybrid bioreactor phase if appropriate, includes micro sampling	
		Fall - CSL site visit/checkup (1 technologist), includes micro sampling	
	2017	Hybrid Bioreactor/CWTS phase, late in operation 2016 or 2017, exact timing to be determined based on results from initial operations of the demo scale and results of hybrid CWTS/bioreactor pilot testing. This will involve adding solid organic matter (such as alfalfa hay, straw) to the CWTS cell(s).	
Reporting	2014-2016	Reporting will be performed twice annually, in the form of an interim update and comprehensive (all data to date) report.	On Schedule

5. Closure

We trust the information herein satisfies your present requirements. Should you have any questions, please contact the persons listed below. We appreciate the opportunity to provide the services detailed in this report, and look forward to discussing any comments you may have.

Respectfully submitted,
Contango Strategies Ltd

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Table A1. Analytical results from initial soil sampled August 28, 2014

Parameter	Units	Rep1	Rep2	Rep3	Rep4	Wood Shavings	Straw	RDL
Calculated Parameters								
Anion Sum	meq/L	5.0	3.1	0.89	0.81	1.0	14	N/A
Cation Sum	meq/L	8.1	6.5	13	5.4	2.2	24	N/A
Cation/EC Ratio	N/A	11	11	10	11	10	9.4	0.10
Ion Balance	N/A	1.6	2.1	15	6.7	2.2	1.7	0.010
Cation exchange capacity	cmol+/Kg	22	20	18	20	<10	50	10
Soluble Parameters								
Soluble Chloride (Cl)	mg/L	41	16	240	26	12	310	5.0
Soluble Conductivity	dS/m	0.72	0.61	1.3	0.51	0.21	2.5	0.020
Soluble (CaCl2) pH	pH	6.46	6.44	6.56	6.47	3.95		N/A
Sodium Adsorption Ratio	N/A	0.55	0.53	2.0	0.65	0.30	0.63	0.10
Soluble Calcium (Ca)	mg/L	110	92	150	71	11	20	1.5
Soluble Magnesium (Mg)	mg/L	14	11	16	8.8	4.0	47	1.0
Soluble Sodium (Na)	mg/L	23	20	97	22	4.6	22	2.5
Soluble Potassium (K)	mg/L	7.6	7.4	8.5	7.1	39	710	1.3
Saturation %	%	47	48	47	44	620	1400	N/A
Soluble Sulphate (SO4)	mg/L	210	180	190	140	<5.0	210	5.0
Nutrients								
Available (NH4F) Nitrogen (N)	mg/kg	8.0	8.1	6.0	5.3	<10	<10	10
Available (NH4F) Phosphorus (P)	mg/kg	5.1	5.7	5.6	3.9	16	200	5.0
Available (NH4OAc) Potassium (K)	mg/kg	55	53	48	61	260	7700	10
Available (CaCl2) Sulphur (S)	mg/kg	29	24	27	24	<20	850	20
Nitrite (N)	mg/L	0.0165	0.0216	0.0156	0.0623	0.0127	<0.50	0.50
Nitrate (N)	mg/L	0.315	0.445	0.332	0.150	0.252	<2.0	2.0
Total Ammonia (N)	mg/L	<0.0050	0.0076	0.0086	0.0090	0.0428	0.067	0.050
Misc. Inorganics								
Fluoride (F)	mg/L	0.220	0.350	0.250	0.210	0.041	0.089	0.010

Parameter	Units	Rep1	Rep2	Rep3	Rep4	Wood Shavings	Straw	RDL
Dissolved Organic Carbon (C)	mg/L	6.55	5.91	4.96	6.85	355	815	50
Total Organic Carbon (C)	%	1.9	1.8	3.1	1.8	48	45	0.10
Alkalinity (Total as CaCO3)	mg/L	33.5	126	36.7	34.8	30.7	304	0.50
Alkalinity (PP as CaCO3)	mg/L	<0.50	13.2	2.49	8.01	<0.50	<0.50	0.50
Bicarbonate (HCO3)	mg/L	40.9	121	38.7	22.9	37.5	371	0.50
Carbonate (CO3)	mg/L	<0.50	15.9	2.99	9.61	<0.50	<0.50	0.50
Hydroxide (OH)	mg/L	<0.50	<0.50	<0.50	<0.50	<0.50	<0.50	0.50
Physical Properties								
% sand by hydrometer	%	59	60	55	57			
% silt by hydrometer	%	26	25	28	27			
Clay Content	%	15	16	17	16			
Texture	N/A	SANDY LOAM	SANDY LOAM	SANDY LOAM	SANDY LOAM			
Total Dissolved Solids	mg/L	80	200	104	222	692	2360	10
Dissolved Metals by ICPMS								
SPLP Aluminum (Al)	ug/L	6540	3640	4840	34400	145	151	2.5
SPLP Antimony (Sb)	ug/L	0.242	0.208	0.243	0.48	0.042	<0.10	0.10
SPLP Arsenic (As)	ug/L	3.24	2.19	2.67	15.0	0.446	1.57	0.10
SPLP Barium (Ba)	ug/L	141	79.9	102	476	46.8	396	0.10
SPLP Beryllium (Be)	ug/L	0.133	0.091	0.142	0.629	0.015	<0.050	0.050
SPLP Bismuth (Bi)	ug/L	0.0440	0.0240	0.0270	0.193	<0.0050	0.039	0.025
SPLP Boron (B)	ug/L	519	233	90	<250	127	<250	250
SPLP Cadmium (Cd)	ug/L	0.0470	0.0410	0.0740	0.158	0.137	0.484	0.025
SPLP Cesium (Cs)	ug/L	0.347	0.206	0.290	1.42	<0.050	0.29	0.25
SPLP Chromium (Cr)	ug/L	10.7	7.80	6.12	43.1	0.62	0.94	0.50
SPLP Cobalt (Co)	ug/L	2.53	1.44	2.13	10.5	0.106	0.291	0.025
SPLP Copper (Cu)	ug/L	248	148	186	608	11.0	116	0.25
SPLP Iron (Fe)	ug/L	8900	4950	7040	47400	49.4	308	5.0
SPLP Lanthanum (La)	ug/L	5.72	4.16	4.81	17.9	<0.050	0.27	0.25

Parameter	Units	Rep1	Rep2	Rep3	Rep4	Wood Shavings	Straw	RDL
SPLP Lead (Pb)	ug/L	3.00	1.79	2.21	12.4	1.10	14.7	0.025
SPLP Lithium (Li)	ug/L	2.90	1.66	2.00	12.6	1.05	3.5	2.5
SPLP Manganese (Mn)	ug/L	125	76.0	103	372	272	597	0.25
SPLP Molybdenum (Mo)	ug/L	1.75	1.46	1.94	1.86	0.230	2.52	0.25
SPLP Nickel (Ni)	ug/L	5.93	3.76	5.39	26.5	0.693	2.22	0.10
SPLP Phosphorus (P)	ug/L	132	84.6	107	503	416	1460	10
SPLP Rubidium (Rb)	ug/L	7.05	4.56	5.97	27.7	13.7	241	0.25
SPLP Selenium (Se)	ug/L	0.487	0.355	0.360	0.69	<0.040	1.09	0.20
SPLP Silicon (Si)	ug/L	14900	8970	11200	67100	425	24000	500
SPLP Silver (Ag)	ug/L	0.123	0.0950	0.123	0.325	0.0560	0.446	0.025
SPLP Strontium (Sr)	ug/L	14.2	12.7	12.2	35.6	11.3	156	0.25
SPLP Tellurium (Te)	ug/L	<0.020	0.024	0.021	<0.10	<0.020	<0.10	0.10
SPLP Thallium (Tl)	ug/L	0.0500	0.0310	0.0410	0.184	0.0110	0.103	0.010
SPLP Thorium (Th)	ug/L	1.44	1.06	1.25	4.56	0.0140	0.151	0.025
SPLP Tin (Sn)	ug/L	0.32	<0.20	0.22	1.1	<0.20	<1.0	1.0
SPLP Titanium (Ti)	ug/L	237	129	176	1280	1.28	4.7	2.5
SPLP Tungsten (W)	ug/L	0.093	0.080	0.086	0.203	0.054	<0.050	0.050
SPLP Uranium (U)	ug/L	0.176	0.134	0.151	0.865	0.0110	0.057	0.010
SPLP Vanadium (V)	ug/L	22.0	11.9	16.2	92.1	1.24	2.3	1.0
SPLP Zinc (Zn)	ug/L	19.9	13.2	15.4	74.4	16.0	84.5	0.50
SPLP Zirconium (Zr)	ug/L	1.72	1.42	1.49	5.77	0.12	<0.50	0.50
SPLP Calcium (Ca)	ug/L	3040	3290	3090	7240	3960	18000	250
SPLP Magnesium (Mg)	ug/L	1420	954	1110	6150	1670	38300	250
SPLP Potassium (K)	ug/L	1250	1010	1440	5780	14100	499000	250
SPLP Sodium (Na)	ug/L	43900	33900	22100	26900	8550	20500	250
SPLP Mercury (Hg)	ug/L	<0.050	<0.050	<0.050	<0.25	0.096	<0.25	0.25
SPLP Sulphur (S)	ug/L	<10000	<10000	<10000	<50000	<10000	<50000	50000

Table A2. Sequentially extracted ICP-MS analytical results from initial soil sampled August 28, 2014

mg/kg	Extraction 1				Extraction 2				Extraction 3				Extraction 4				Extraction 5				RDL
	R1	R2	R3	R4	R1	R2	R3	R4	R1	R2	R3	R4	R1	R2	R3	R4	R1	R2	R3	R4	
Al	-	-	-	-	34	28	30	34	800	640	890	720	920	1500	1300	1300	17000	17000	16000	15000	10
Sb	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	2.0
As	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	6.5	6.7	6.7	6.3	2.0
Ba	45	24	39	46	91	51	80	76	120	100	120	110	46	150	63	65	540	620	510	530	5.0
Be	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	2.0
B	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	5.0
Cd	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	0.30
Cr	-	-	-	-	-	-	-	-	-	-	-	-	2.5	6.2	3.4	3.5	21	31	19	19	2.0
Co	-	-	-	-	-	-	-	-	2.6	2.4	2.4	2.2	-	-	-	-	12	11	11	11	1.0
Cu	2.9	4.9	2.8	3.7	93	35	67	120	270	180	240	290	480	380	630	650	1500	960	1400	1300	2.0
Fe	-	-	-	-	-	85	-	-	5000	4400	5300	4700	340	610	440	550	35000	31000	34000	33000	50
Pb	-	-	-	-	-	-	-	-	1.4	1.3	1.5	1.4	-	1.1	0.60	0.74	6.0	8.4	5.3	5.3	0.50
Mn	-	-	-	2.3	140	76	120	130	160	110	150	150	19	21	23	28	750	490	670	700	2.0
Mo	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	2.9	2.0
Ni	-	-	-	-	-	-	-	-	3.7	5.9	3.7	3.4	-	3.3	-	-	18	28	17	17	2.0
Se	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	2.0
Ag	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	0.50
Ar	14	13	12	14	7.7	7.3	6.9	6.2	-	5.8	-	-	-	-	-	-	48	58	45	47	5.0
Tl	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	0.22	0.16	0.20	0.20	0.10
U	-	-	-	-	0.19	0.46	0.17	0.15	0.17	0.38	0.17	0.15	-	0.45	0.10	0.11	0.87	2.1	0.78	0.77	0.10
V	-	-	-	-	-	-	-	-	10	12	11	9.3	-	3.8	-	-	80	79	75	74	2.0
Zn	-	-	-	-	-	-	-	-	11	9.1	12	10	-	6.1	5.2	5.5	89	70	83	82	5.0

Table A3. Sequentially extracted ICP-MS analytical results from initial wood shavings and straw amendments sampled August 28, 2014

mg/kg	Wood Shavings					Straw					RDL
	E1	E 2	E3	E4	E5	E1	E2	E3	E4	E5	
Al	-	-	-	-	-	-	-	-	-	35	10
Sb	-	-	-	-	-	-	-	-	-	-	2.0
As	-	-	-	-	-	-	-	-	-	-	2.0
Ba	-	-	-	-	10	-	11	10	-	55	5.0
Be	-	-	-	-	-	-	-	-	-	-	2.0
B	-	-	-	-	-	-	-	-	-	-	5.0
Cd	-	-	-	-	-	-	-	-	-	-	0.30
Cr	-	-	-	-	-	-	-	-	-	-	2.0
Co	-	-	-	-	-	-	-	-	-	-	1.0
Cu	-	-	-	-	-	-	-	-	-	3.5	2.0
Fe	-	-	-	-	-	-	-	-	-	140	50
Pb	-	-	-	-	-	-	-	-	-	2.6	0.50
Mn	21	13	5.4	-	86	3.5	14	7.5	-	70	2.0
Mo	-	-	-	-	-	-	-	-	-	-	2.0
Ni	-	-	-	-	-	-	-	-	-	7.1	2.0
Se	-	-	-	-	-	-	-	-	-	-	2.0
Ag	-	-	-	-	-	-	-	-	-	-	0.50
Ar	-	-	-	-	-	-	-	-	-	13	5.0
Tl	-	-	-	-	-	-	-	-	-	-	0.10
U	-	-	-	-	-	-	-	-	-	-	0.10
V	-	-	-	-	-	-	-	-	-	-	2.0
Zn	-	-	-	-	9.0	-	-	-	-	15	5.0

Table A4. Analytical results from plants sampled August 27, 2014

Parameter	Units	Carex						Moss			RDL
		Roots			Leaves						
		Rep 1	Rep 2	Rep 3	Rep 1	Rep 2	Rep 3	Rep 1	Rep 2	Rep 3	
Nutrients											
Available Nitrogen (N)	mg/kg	<10	<10	<10	<10	<10	<10	<10	<10	<10	10
Available Phosphorus (P)	mg/kg	240	48	91	640	520	810	470	340	150	5.0
Available Potassium (K)	mg/kg	8100	1400	3200	9100	8100	11000	4000	5400	2700	10
Available Sulphur (S)	mg/kg	210	670	700	170	340	530	290	400	220	20
Misc. Inorganics											
Total Organic Carbon (C)	%	39	36	37	38	38	43	43	45	40	0.10
Total Carbon	%	41	37	38	45	40	44	41	45	44	0.20
Inorganics											
Moisture	%	87	88	86	74	77	75	90	94	92	1.0
Metals											
Antimony (Sb)	mg/kg	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	0.05
Arsenic (As)	mg/kg	1.7	1.0	0.4	0.3	0.1	<0.1	0.2	0.1	<0.1	0.1
Barium (Ba)	mg/kg	53.9	34.3	28.2	31.0	23.3	17.9	29.9	22.3	10.5	0.3
Beryllium (Be)	mg/kg	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	0.05
Bismuth (Bi)	mg/kg	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	0.05
Boron (B)	mg/kg	<0.5	0.6	0.8	1.0	0.8	1.1	0.5	<0.5	<0.5	0.5
Cadmium (Cd)	mg/kg	0.02	<0.01	0.01	<0.01	<0.01	<0.01	0.01	<0.01	0.01	0.01
Calcium (Ca)	mg/kg	973	648	752	1160	944	743	1460	1210	1160	50
Chromium (Cr)	mg/kg	1.4	1.2	0.6	1.3	0.5	0.4	<0.3	<0.3	<0.3	0.3
Cobalt (Co)	mg/kg	1.45	0.864	0.965	0.594	0.333	0.164	0.645	0.464	0.166	0.005
Copper (Cu)	mg/kg	93.7	47.0	15.1	20.5	21.9	6.8	13.8	18.3	11.2	0.5
Iron (Fe)	mg/kg	7870	4540	2560	2080	1140	540	1190	1020	404	3
Lead (Pb)	mg/kg	0.27	0.11	0.11	0.08	0.06	<0.03	0.04	0.04	0.03	0.03
Magnesium (Mg)	mg/kg	910	477	327	653	321	325	293	229	190	100
Manganese (Mn)	mg/kg	98.3	128	151	116	282	141	272	179	27.1	0.3
Molybdenum (Mo)	mg/kg	0.18	0.21	0.15	0.31	0.41	0.17	0.29	0.09	0.25	0.05
Nickel (Ni)	mg/kg	0.87	0.41	0.37	0.32	0.20	0.11	0.36	0.49	0.13	0.05
Phosphorus (P)	mg/kg	<500	<500	<500	<1000	<1000	<1000	<500	<500	<500	500
Potassium (K)	mg/kg	1180	1070	1440	2860	3620	3430	674	363	315	100
Selenium (Se)	mg/kg	<0.2	<0.2	<0.2	<0.2	<0.2	<0.2	<0.2	<0.2	<0.2	0.2
Silver (Ag)	mg/kg	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	<0.05	0.05
Sodium (Na)	mg/kg	509	512	508	222	149	180	94	54	<50	50
Strontium (Sr)	mg/kg	5.8	4.4	5.7	5.7	5.4	4.7	10.1	8.2	7.1	0.5
Thallium (Tl)	mg/kg	0.007	<0.003	<0.003	<0.003	<0.003	<0.003	<0.003	<0.003	<0.003	0.003
Tin (Sn)	mg/kg	<0.3	<0.3	<0.3	<0.3	<0.3	<0.3	<0.3	<0.3	<0.3	0.3
Titanium (Ti)	mg/kg	109	49	24.3	42	22.8	8.0	6.0	5.9	4.6	0.5
Uranium (U)	mg/kg	0.049	0.025	0.035	0.016	0.016	0.012	0.014	0.015	0.007	0.005
Vanadium (V)	mg/kg	11.8	4.9	3.2	3.5	1.8	0.7	0.7	0.8	0.4	0.3
Zinc (Zn)	mg/kg	12	6	4	7	7	4	5	4	2	2

Table A5. Analytical results from water sampled September 18, 2014

Parameter	Units	CWTS LINE 1	CWTS 1A	CWTS 1B	CWTS LINE 2	CWTS 2A	CWTS 2B	CWTS FINAL	RDL
Elements									
Total Aluminum (Al)	mg/L	0.13	0.031	0.044	0.011	0.021	0.016	0.026	0.0030
Dissolved Aluminum (Al)	mg/L	0.0062	0.0093	0.015	0.0091	0.0097	0.013	0.017	0.0030
Total Antimony (Sb)	mg/L	<0.00060	<0.00060	<0.00060	<0.00060	<0.00060	<0.00060	<0.00060	0.00060
Dissolved Antimony (Sb)	mg/L	<0.00060	<0.00060	<0.00060	<0.00060	<0.00060	<0.00060	<0.00060	0.00060
Total Arsenic (As)	mg/L	0.00041	0.00041	0.00044	0.00034	0.00039	0.00038	0.00039	0.00020
Dissolved Arsenic (As)	mg/L	0.00038	0.00038	0.00040	0.00034	0.00034	0.00039	0.00042	0.00020
Total Barium (Ba)	mg/L	0.13	0.13	0.13	0.13	0.13	0.13	0.13	0.010
Dissolved Barium (Ba)	mg/L	0.12	0.12	0.13	0.12	0.12	0.13	0.13	0.010
Total Beryllium (Be)	mg/L	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	0.0010
Dissolved Beryllium (Be)	mg/L	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	0.0010
Total Boron (B)	mg/L	<0.020	<0.020	<0.020	<0.020	0.024	0.020	<0.020	0.020
Dissolved Boron (B)	mg/L	<0.020	<0.020	<0.020	<0.020	<0.020	<0.020	<0.020	0.020
Total Cadmium (Cd)	ug/L	0.048	0.045	0.039	0.046	0.040	0.029	0.021	0.020
Dissolved Cadmium (Cd)	ug/L	0.046	0.036	0.030	0.041	0.038	0.022	<0.020	0.020
Total Calcium (Ca)	mg/L	130	130	130	130	130	130	130	0.30
Dissolved Calcium (Ca)	mg/L	130	120	120	130	120	120	120	0.30
Total Chromium (Cr)	mg/L	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	0.0010
Dissolved Chromium (Cr)	mg/L	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	0.0010
Total Cobalt (Co)	mg/L	<0.00030	<0.00030	<0.00030	<0.00030	<0.00030	<0.00030	<0.00030	0.00030
Dissolved Cobalt (Co)	mg/L	<0.00030	<0.00030	<0.00030	<0.00030	<0.00030	<0.00030	<0.00030	0.00030
Total Copper (Cu)	mg/L	0.088	0.047	0.046	0.053	0.047	0.041	0.041	0.00020
Dissolved Copper (Cu)	mg/L	0.046	0.041	0.039	0.047	0.042	0.036	0.034	0.00020
Total Iron (Fe)	mg/L	0.50	0.20	0.20	0.19	0.18	0.15	0.15	0.060
Dissolved Iron (Fe)	mg/L	0.13	0.11	0.12	0.15	0.12	0.10	0.10	0.060
Total Lead (Pb)	mg/L	<0.00020	<0.00020	<0.00020	<0.00020	<0.00020	<0.00020	<0.00020	0.00020
Dissolved Lead (Pb)	mg/L	<0.00020	<0.00020	<0.00020	<0.00020	<0.00020	<0.00020	<0.00020	0.00020

Parameter	Units	CWTS LINE 1	CWTS 1A	CWTS 1B	CWTS LINE 2	CWTS 2A	CWTS 2B	CWTS FINAL	RDL
Total Lithium (Li)	mg/L	<0.020	<0.020	<0.020	<0.020	<0.020	<0.020	<0.020	0.020
Dissolved Lithium (Li)	mg/L	<0.020	<0.020	<0.020	<0.020	<0.020	<0.020	<0.020	0.020
Total Magnesium (Mg)	mg/L	37	38	37	38	37	37	38	0.20
Dissolved Magnesium (Mg)	mg/L	37	36	37	37	36	36	36	0.20
Total Manganese (Mn)	mg/L	0.26	0.19	0.16	0.26	0.19	0.14	0.094	0.0040
Dissolved Manganese (Mn)	mg/L	0.25	0.18	0.15	0.25	0.20	0.14	0.090	0.0040
Total Molybdenum (Mo)	mg/L	0.0083	0.0082	0.0083	0.0084	0.0083	0.0081	0.0082	0.00020
Dissolved Molybdenum (Mo)	mg/L	0.0079	0.0076	0.0078	0.0077	0.0078	0.0079	0.0076	0.00020
Total Nickel (Ni)	mg/L	0.0013	0.0012	0.0015	0.0012	0.0012	0.0011	0.0012	0.00050
Dissolved Nickel (Ni)	mg/L	0.00075	0.00079	0.00079	0.00072	0.00074	0.00081	0.00068	0.00050
Total Phosphorus (P)	mg/L	0.12	<0.10	<0.10	<0.10	0.14	0.11	<0.10	0.10
Dissolved Phosphorus (P)	mg/L	<0.10	0.13	0.12	<0.10	<0.10	<0.10	<0.10	0.10
Total Potassium (K)	mg/L	4.5	4.5	4.4	4.5	4.5	4.4	4.6	0.30
Dissolved Potassium (K)	mg/L	4.3	4.3	4.4	4.3	4.3	4.3	4.4	0.30
Total Selenium (Se)	mg/L	0.0064	0.0065	0.0064	0.0066	0.0063	0.0063	0.0062	0.00020
Dissolved Selenium (Se)	mg/L	0.0063	0.0061	0.0061	0.0063	0.0061	0.0063	0.0059	0.00020
Total Silicon (Si)	mg/L	7.8	7.2	7.0	7.6	7.2	7.1	7.2	0.10
Dissolved Silicon (Si)	mg/L	7.3	6.9	6.8	7.3	6.9	6.9	6.8	0.10
Total Silver (Ag)	mg/L	<0.00010	<0.00010	<0.00010	<0.00010	<0.00010	<0.00010	<0.00010	0.00010
Dissolved Silver (Ag)	mg/L	<0.00010	<0.00010	<0.00010	<0.00010	<0.00010	<0.00010	<0.00010	0.00010
Total Sodium (Na)	mg/L	22	22	22	22	22	22	23	0.50
Dissolved Sodium (Na)	mg/L	22	21	22	22	21	22	22	0.50
Total Strontium (Sr)	mg/L	1.3	1.3	1.3	1.4	1.3	1.3	1.3	0.020
Dissolved Strontium (Sr)	mg/L	1.3	1.3	1.3	1.3	1.3	1.3	1.3	0.020
Total Sulphur (S)	mg/L	44	45	43	44	43	44	44	0.20
Dissolved Sulphur (S)	mg/L	42	41	42	42	41	42	42	0.20
Total Thallium (Tl)	mg/L	<0.00020	<0.00020	<0.00020	<0.00020	<0.00020	<0.00020	<0.00020	0.00020
Dissolved Thallium (Tl)	mg/L	<0.00020	<0.00020	<0.00020	<0.00020	<0.00020	<0.00020	<0.00020	0.00020

Parameter	Units	CWTS LINE 1	CWTS 1A	CWTS 1B	CWTS LINE 2	CWTS 2A	CWTS 2B	CWTS FINAL	RDL
Total Tin (Sn)	mg/L	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	0.0010
Dissolved Tin (Sn)	mg/L	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	0.0010
Total Titanium (Ti)	mg/L	0.014	0.0023	0.0038	<0.0010	0.0021	<0.0010	0.0015	0.0010
Dissolved Titanium (Ti)	mg/L	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	0.0010
Total Uranium (U)	mg/L	0.0031	0.0029	0.0031	0.0031	0.0031	0.0030	0.0031	0.00010
Dissolved Uranium (U)	mg/L	0.0032	0.0032	0.0032	0.0032	0.0032	0.0034	0.0032	0.00010
Total Vanadium (V)	mg/L	0.0011	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	0.0010
Dissolved Vanadium (V)	mg/L	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	<0.0010	0.0010
Total Zinc (Zn)	mg/L	0.075	0.055	0.049	0.071	0.059	0.042	0.029	0.0030
Dissolved Zinc (Zn)	mg/L	0.068	0.049	0.037	0.068	0.052	0.038	0.022	0.0030
Misc.									
Dissolved Nitrite (N)	mg/L	0.016	0.020	0.020	0.016	0.020	0.022	0.026	0.010
Dissolved Nitrate (N)	mg/L	12	12	12	12	12	12	12	0.020
Total Ammonia (N)	mg/L	<0.050	<0.050	<0.050	<0.050	<0.050	<0.050	<0.050	0.050
Total Nitrogen (N)	mg/L	13	13	13	13	13	13	13	0.050
Biochemical Oxygen Demand	mg/L	<2.0	<2.0	<2.0	<2.0	<2.0	<2.0	<2.0	2.0
Total Chemical Oxygen Demand	mg/L	27	28	31	26	41	27	27	5.0
Conductivity	uS/cm	920	920	910	930	920	910	900	1.0
pH	pH	8.27	8.23	8.34	8.20	8.32	8.30	8.44	N/A
Total Organic Carbon (C)	mg/L	9.7	9.3	9.7	9.5	10	11	10	0.50
Total Suspended Solids	mg/L	0.67	1.1	1.3	<0.40	0.53	0.67	0.40	0.40
Alkalinity (PP as CaCO3)	mg/L	<0.50	<0.50	0.90	<0.50	<0.50	<0.50	5.0	0.50
Alkalinity (Total as CaCO3)	mg/L	300	300	300	300	300	300	290	0.50
Bicarbonate (HCO3)	mg/L	360	360	360	370	360	360	350	0.50
Carbonate (CO3)	mg/L	<0.50	<0.50	1.1	<0.50	<0.50	<0.50	6.0	0.50
Hydroxide (OH)	mg/L	<0.50	<0.50	<0.50	<0.50	<0.50	<0.50	<0.50	0.50
Dissolved Sulphate (SO4)	mg/L	140	140	140	140	140	140	140	1.0
Dissolved Chloride (Cl)	mg/L	22	21	22	21	22	21	21	1.0

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Minto Mine Constructed Wetland Treatment Research Program – Demonstration Scale 2015 Update

Document – 011_0316_03B

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1. Introduction and Background

The Minto Mine, operated by Capstone Mining Corp., is located 240 km northwest of Whitehorse on the west side of the Yukon River. The Minto property lies within the eastern part of the Dawson Range, with elevations from 700 to 1,000 m; the landscape has rounded mountains intersected by broad valleys and drainages that are part of the Yukon River watershed.

The Minto Mine has been in commercial operation since October 2007 and the deposits being mined are copper sulphide mineralized zones. Surface and groundwater water quality is a key consideration in the evaluation of potential effects of mining and mineral development projects and changes to water quality parameters have the potential to affect aquatic and human use of water resources. A Reclamation and Closure Plan (RCP) is required under both the Water Licence and the Quartz Mining Licence. The RCP is intended to address the long-term physical and chemical stability of the site and closure of the proposed features and disturbances associated with the mine. As a part of the RCP, a Constructed Wetland Treatment System (CWTS) is being designed, evaluated, and optimized for water treatment at closure through a phased program (Minto Phase V/VI Expansion Project, YOR Project Number 2013-0100).

In order for CWTSs to be effective, they must be designed, piloted, optimized, implemented, and maintained in a site-specific manner. A scaled approach for CWTS implementation allows for improvement, optimization, and flexibility for modifications along each step. Phases include: 1) site assessment and information gathering, 2) technology selection and conceptual design, 3) pilot-scale testing and optimization (controlled environment), 4) on-site demonstration-scale confirmation and optimization, and 5) full-scale implementation. Phases 1-3 have been completed (reports 2013-0100-256 and 2013-0100-257 on YESAB registry, and Contango, March 2014; Contango, November 2014) and confirmed plant amenability to transplantation and the CWTS design for further on-site testing. During pilot-scale trials, the selected CWTS design achieved on average 92% removal of copper (mean influent 146 µg/L, outflow 11.3 µg/L) and 41% removal of selenium (mean influent 10.2 µg/L, outflow 6 µg/L) using synthetic influent designed to mimic the worst-case water chemistry of a long-term closure scenario, but tested under the controlled conditions (e.g., controlled flows, known temperature, etc) of an off-site treatability testing center (Contango, November 2014). It should be noted that lower influent concentrations will have a lower percent removal even when achieving the same final outflow concentrations.

Phase 4 of the project is underway, with the on-site demonstration scale CWTS constructed at the Minto Mine during fall 2014 (Contango, March 2015). This document reports on the on-site demonstration scale CWTS data from construction through 2015.

2. Construction

2.1. System layout and dimensions

The demonstration-scale CWTS includes 2 systems in parallel with 2 cells in each series and a final catchment basin that both systems flow into (Figure 1). Dimensions and construction details are available in the Minto Demonstration Scale Report Document 011_0315_01A (Contango, March 2015). The two parallel systems serve as a replicate for data analysis, and as testing has progressed, the two systems have also allowed for comparison of different management techniques. Dimensions of the systems are provided here in Table 1.

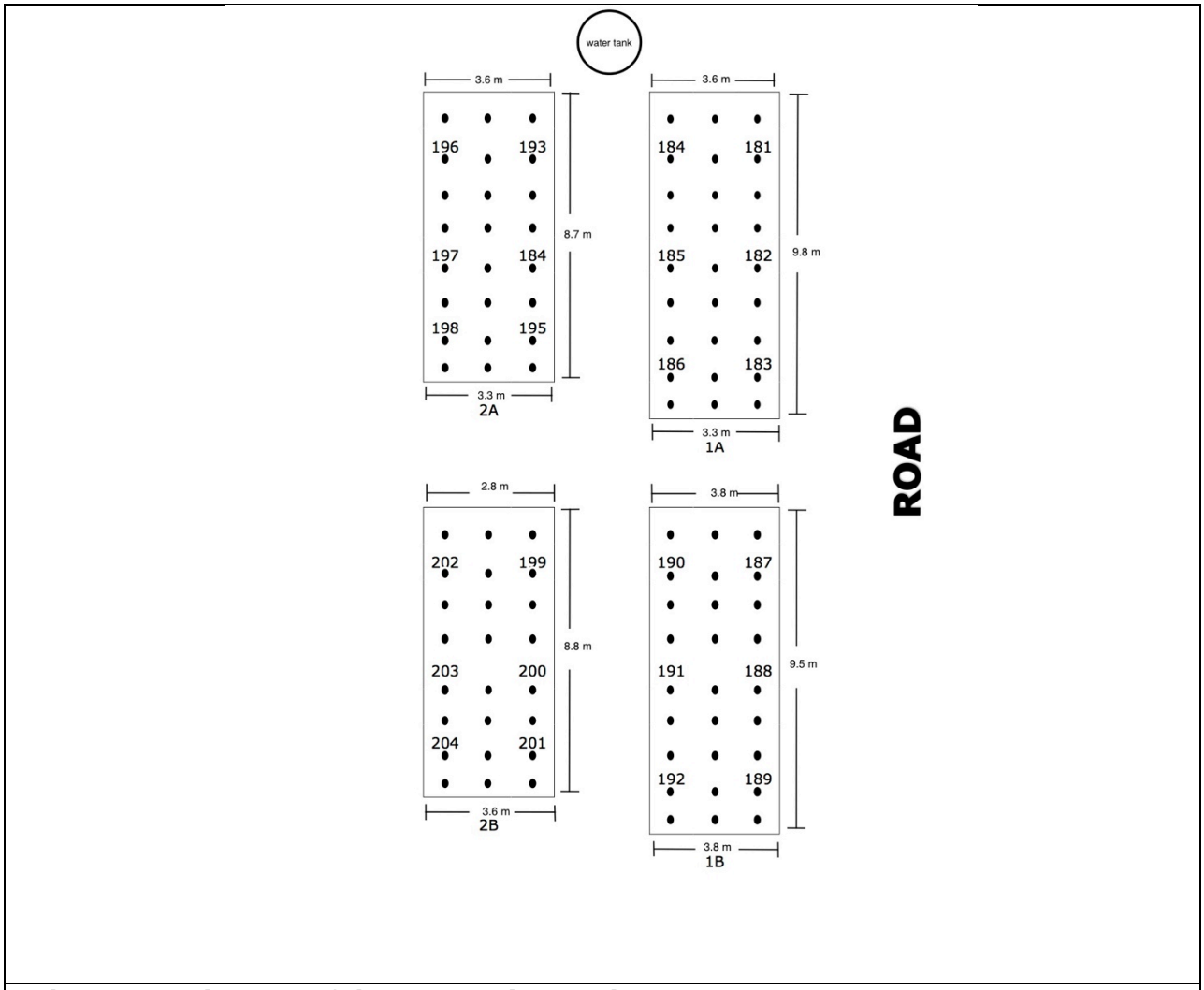


Figure 1 – Diagram of demonstration-scale CWTS.
Dimension measurements are indicated at soil surface. Black dots indicate initial construction grid marked by moss stakes, and locations of soil redox probes with identifying numbers.

Table 1. Measurements of demonstration-scale CWTS cells at soil surface and resultant areas of treatment systems.

Measurement		Series 1		Series 2	
		1A	1B	2A	2B
Width (m)	Inflow	3.6	3.8	3.6	2.8
	Outflow	3.3	3.8	3.3	3.6
Length (m)		9.8	9.5	8.7	8.8
Approximate surface area at soil (m^2)		33.8	36.1	30.0	28.2
Total area of system at soil (m^2)		69.9		58.2	

2.2.Substrate Soils in Construction

Soils used to construct the CWTS are described in the initial report that outlines construction (Contango, March 2015). In brief, the recommended soil for the CWTS was sand, with 2-7% by volume as organic material (e.g., woodchips, peat). In the pilot-scale systems, this resulted in a total organic carbon (TOC) content by weight of 0.2-0.6% (the sand itself was at 0.1% TOC prior to adding amendment). Ideally, this number could be higher, approximately 2-10% by weight to stimulate the desired reducing conditions. For the demonstration-scale system, the soil added to each of the 4 cells during construction was from a local borrow site. As is expected in a mining area, the soils are likely mineralized. Although a potential borrow source was tested prior to construction ("Tested Soil"; Table 2), a different borrow source was available upon construction of the demonstration-scale wetland ("Soil Used"; Table 2). The material used in the construction of the demonstration-scale wetland was an organic peat, and analyses received after construction indicated an elevated concentration of leachable copper (Table 2). It should be noted that in a full-scale system, the variability in soil substrate would be normalized by the larger volume of soil used. The substrate in the wetland (composed of the organic peat soil, wood chips, and straw) had a TOC content of 1.8-3.1%. The implications of this are discussed in Section 4 if this report.

Table 2. Comparison of Copper Concentrations in Soils of Pilot- and Demonstration-Scale CWTS

Test Method	Pilot-scale	Demonstration-scale	
	Initial Soil	Tested Soil (June, 2014)	Soil Used (August, 2014)
SPLP Copper (mg/L)[1]	-	0.00546-0.0296[2]	0.148-0.608
Total Copper (mg/kg)	5.3-5.5	210-1400[2]	960-1400[3]

[1] SPLP - Synthetic Precipitation Leaching Procedure
[2] For the June 2014 samples, the soil with the highest total copper concentration also had the lowest leachable copper concentration, and was therefore deemed acceptable for use.
[3] Total copper value for soils used was taken in June 2015 (no data for August 2014).

2.3. Vegetation Used in Planting

The demonstration-scale CWTS was planted with *Carex aquatilis* (aquatic sedge) and aquatic mosses from the W10 area of the Minto Site. The plant selection and borrow source was previously determined through the site assessment (reports 2013-0100-256 and 2013-0100-257 on the YESAB registry) and pilot-scale testing (Contango, November 2014). Five *C. aquatilis* were planted per square meter, with moss tied to stakes that outlined the 1 m x 1 m grid for planting (details provided in Contango, March 2015).

2.4. Water source

Water from the W36 area receiving seepage from the toe of the Mill Valley Fill Extension (MVFE) was selected for the demonstration-scale CWTS testing as the leachate is similar to that expected upon closure in the MVFE area. The chemistry of this water at the time of bringing the demonstration-scale CWTS online (September 18, 2014) is provided in Contango, March 2015.

After construction, the flow rates for the systems were planned to be set to have a 10 day nominal hydraulic retention time (HRT), meaning water entering the wetland takes 10 days to exit. This is referred to as a nominal HRT because it is a calculation based on the size of the wetland and the amount of water entering, and not confirmed empirically using tracing dyes. This HRT is much longer than what is necessary to achieve treatment based on the pilot-scale systems (~3 days), and was chosen to facilitate plant establishment and maturation. However, due to the leachable copper concentrations in the soils used for construction, it was instead decided to run the systems at a faster flow to wash as much leachable copper from the soils as possible. As such, a shorter nominal HRT of ~20 hrs was used as a starting point (HRT calculated using measured water depths of approximately 15 cm and negligible pore water involvement [30cm of soil at 10% pore volume] due to peat soils). The actual HRT will be confirmed by a tracer study in 2016 (Section 8). Despite the shorter HRT resulting from the faster flow of water, the systems have acclimated and matured as was expected for the longer HRT. Because the sizes of the CWTS systems are slightly different, Series 1 (closer to the road) and Series 2 (further from road) are set with flow rates to result in similar HRTs in each system. Flow rates were monitored and adjusted throughout 2015 based on CWTS establishment and maturation (Sections 3 and 4).

3. Commissioning

The time period between the construction of the CWTS and achieving the expected treatment performance is referred to as the commissioning period. This period is needed for operational adjustments to be made (e.g., raising water depth, modifying outflow patterns), and for plant and microbial populations to establish and mature. Based on pilot-scale testing (Contango, November 2014), the estimated commissioning period for the Minto demonstration-scale CWTS is 4 months of operation (i.e., period when water is flowing). In 2014, the demonstration-scale CWTS operated for less than one month prior to freezing, and was restarted along with spring thaw (freshet) in mid-May 2015. Based on these timelines, September 2015 was month 4 of commissioning, and the demonstration-scale system is trending towards performance as expected during the commissioning period.

Functionality similar to that of the matured pilot-scale systems is expected to occur by summer 2016.

Operational adjustments made in 2015 include:

- Increasing water depth
- Adding more aquatic moss
- Modifying inflow and outflow distribution
- Removing *Carex utriculata* (a small quantity was erroneously planted)
- Wrapping inflow tank to prevent algae growth

While full treatment functionality of the demonstration-scale system is not expected during the commissioning period, there is important data that should be collected during this period as it applies to the full-scale system in the following ways:

- evaluation of construction effectiveness and optimizations that could be made.
- assessment of timelines to reach targeted operational parameters to allow for effective phasing of implementation (e.g., soil redox, percentage and abundance of sulphate reducing bacteria).
- Effectiveness of *Carex aquatilis* and aquatic moss transplantation to assess planting density, time period to full density, and if plant propagation and/or a replanting schedule is necessary.

4. Timeline and Sampling Schedule

Major events and operational adjustments are listed in Table 3. The actual dates as well as relative days of operation are provided, which adjusts for time that the CWTS was not receiving water as it was frozen. The relative days of operation allows for comparison to expected timelines from the pilot-scale testing, and for planning and scheduling to be done for full-scale construction and commissioning in the context of mine closure planning.

The sampling schedule for 2015 was conceptually developed prior to beginning construction of the demonstration-scale CWTS (Appendix A). Actual dates of sampling were dependent on timing of spring thaw and winter freeze-up, and the associated ability to have the pumps operating at the W36 pond to supply water to the demonstration-scale CWTS.

Table 3. Events and sampling activities since construction

Event	Key Activity	Flow Rate Setting m^3/day (gal/min)		Calendar Date	Day of Operation
		CWTS Series 1	CWTS Series 2		
CWTS constructed and planted	First sampling, water started.	-	-	August 27 – 31, 2014	0-4
Freeze up for winter	Feed water pumps turned off.	-	-	September 19, 2014	23
Start up for 2015	Feed water pumps turned on.	14.17 (2.60)	11.61 (2.13)	May 16, 2015	24
Contango Site Visit #1	Microbiology, soils, water tested. *Carex* stem counts. Added more aquatic moss. Put black wrap on water tank to prevent algal growth. Water depth adjusted with sandbags.			June 18, 2015	57
Flow rate increased	Flow rates increased.	17.44 (3.20)	15.81 (2.90)	July 13, 2015	82
Contango Site Visit #2	Microbiology, soils, water tested. Added more aquatic moss.			August 16, 2015	116
Contango Site Visit #3	Water tested. Started Fe-EDTA test on System 2.			September 17, 2015	148
	Microbiology, soils, plants, water tested.			September 18, 2015	149
Fe-EDTA Testing	Daily total and dissolved copper analysis conducted at Minto.			September 19 – 26, 2015	150-157
Freeze up for Winter	Feed water pumps turned off.	-	-	September 29, 2015	160

5. Monitoring Explanatory Parameters

Explanatory parameters are quantifiable aspects of a CWTS environment that can be used to assess feasibility of treatment for a range of constituents, and therefore 'explain' the performance of a CWTS. These parameters, which often include acidity, alkalinity, conductivity, dissolved oxygen (DO), pH, oxidation reduction potential (ORP), ion balance, available electrons donors (e.g., organic carbon, reduced elements), and temperature, can be used to predict, promote, and/or optimize the ability of the system to treat different constituents (Haakensen et al., 2015). A YSI ProPlus meter was used in the field to test for water temperature, DO, conductivity (and specific conductivity; SPC), pH, and ORP.

Average water temperature of the demonstration-scale CWTS was 12.9°C, ranging from 0.5°C to 25°C through 2015. As would be expected, both the month of testing and time of day were found to affect temperature variation. DO concentrations of the CWTS feed and cells were on average 10 mg/L, which is higher than the pilot-scale systems (average 4.8 mg/L), which make it more difficult to carry out nitrate and selenium treatment, and sulphide production for copper and cadmium treatment. The conductivity, pH and ORP of the demonstration-scale CWTS cells are all very similar to the pilot-scale systems.

An additional explanatory parameter that can be used to monitor maturation of the wetland during the commissioning period is the soil redox potential, which is measured using platinum tip probes (in soil) and Calomel electrodes (in water). This measurement offers insight into the direction of electron flux between the sediment/soil/pore water and overlying water column (Faulkner et al., 1989; Huddleston & Rodgers, 2008), and can be used to confirm reducing conditions in the soil. Based on the information gathered in pilot-scale testing, the targeted soil redox for the demonstration-scale CWTS is between -100 and -250 mV. In these redox ranges, bacterial sulphide-production through reduction of sulphur compounds (e.g., sulphate) is expected. Sulphide production directly results in metals and metalloid treatment for elements such as cadmium, copper, molybdenum, nickel, lead, and zinc by precipitation as metal sulphides (Figure 2).

This maturation period is necessary for sufficient quantities of microbes to populate the wetland and become active in decomposing organic material. It is the electrons produced by the decomposition of organic material that is reported by the soil redox measurements. The decomposition of organic material then feeds the sulphate-reducing bacteria the type of energy they need to produce the sulphides that remove the copper from the water. The microbial activity of the system is discussed further in Section 6.5.

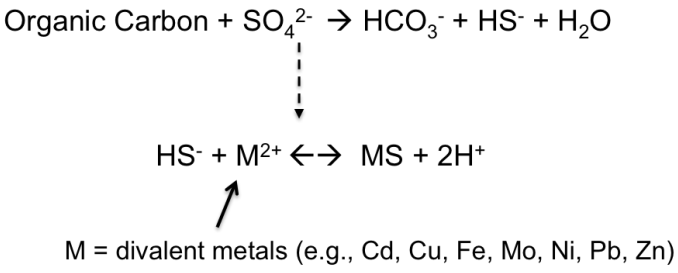


Figure 2 – Example equation of sulphate reduction
Equation is not balanced; demonstrates sulphate reduction with creation of alkalinity and sulphides, which can then form metal sulphide complexes.

As expected from the pilot-scale testing, the soil redox in all of the demonstration-scale CWTS cells decreased over time, indicating maturation of the system (Figure 2). At the end of 2015, only Series 1 had begun to achieve targeted soil redox values that are conducive to sulphide production (ahead of anticipated schedule). It is possible that Series 1 had more organic material in the soils than Series 2, because of how construction occurred and this wasn't reflected in the small sample size sent for analytical testing.

Some soil redox probes were reporting negative values in the targeted range by July 18, 2015 (87 days); however, in general the CWTS took approximately 4 months of operation to establish generally reducing conditions. By September 17, 2015 (day 148), most of the soil redox probes within Series 1 were reporting redox values within the targeted performance range, with Series 2 still trending downwards but not yet within the targeted range. In comparison, the pilot-scale systems were stable and reducing within approximately 4 months of construction (Figure 2).

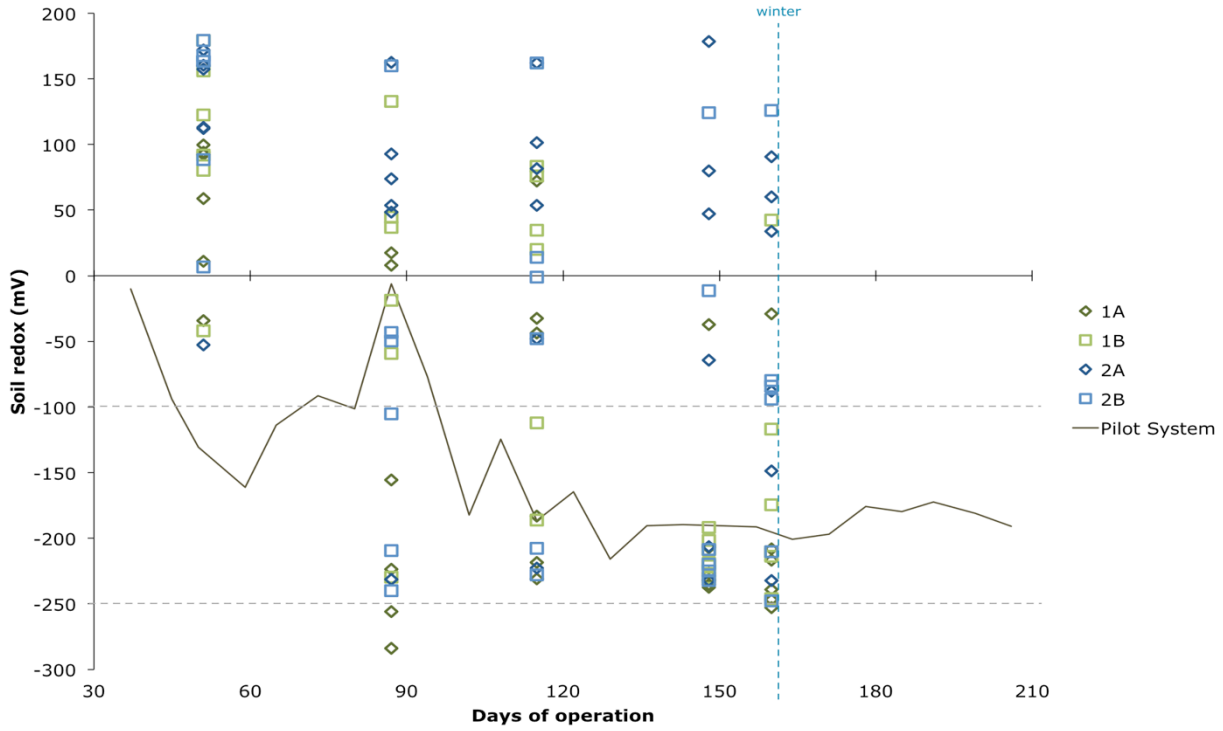


Figure 3 – Soil redox potential of each CWTS cell over time

All demonstration-scale CWTS soil redox measurements are plotted, along with the average soil redox of the pilot-scale systems. Targeted soil redox values based on pilot-scale testing are indicated with dotted lines. The blue dotted line indicates winter 2015.

6. Performance progress during commissioning period

6.1. Soils

To assess the effect of copper in the soils on CWTS functionality, three soil analytical test methods were used:

- total concentration of elements in the soils by ICP
- leachable concentrations of elements in the soils by Synthetic Precipitation Leaching Procedure (SPLP), which is a method to assess the mobility of elements in soils at the pH of rain water (i.e., if the wetland were to entirely dry out, then be subjected to leaching by rain water)
- concentration of elements in chemical extractions by sequential ICP-MS to assess the stability and form of elements in soils (Table 4)

Table 4. Summary of extractable fractions from sequential ICP-MS analysis [1]

Fraction	Description	Elements Unstable when
1	Exchangeable fraction for adsorbed minerals	Readily released (i.e., soluble and exchangeable)
2	Mineral fraction bound to carbonates	Decreased pH
3	Mineral fraction bound to Fe-Mn oxides	Reducing conditions
4	Mineral fraction bound to organic matter and sulphides	Oxidizing conditions
5	Residual mineral fraction (primary and secondary minerals)	Not expected to be released in solution over time under conditions normally encountered in nature

[1] Method based on Tessier *et al.,* 1979

Although unintentional, the use of soils in the CWTS with high initial leachable copper concentrations (Table 2) allowed for additional types of testing to be carried out on these systems. Because the soil substrates used for construction of the CWTS were from overburden sources, the copper was not in a mineral form that would typically be found in a reducing CWTS (i.e., negative relative soil redox). Therefore, there has been some initial leaching of copper (and other elements, such as aluminum) from the soils into the water (Figure 5 and Figure 7).

While the total copper concentrations have not changed significantly, the leachable copper (measured by SPLP) has decreased over time. By day 58 of operation, the top 5 cm of soil was decreased to only 10-20% of the initial leachable copper concentration, and the lower 10-20 cm layers reached similar concentrations by day 149 (Table 5). Analysis of the soils by ICP-MS with sequentially extracted acid analysis shows that despite elevated initial leachable copper concentrations, the soils have become more stable (less leachable) over time in the wetland setting as the soils have aged (Figure 4). This beneficial aging of soils to less soluble (sulphide) mineralized form is expected for this type of treatment wetland

design. It should be noted that due to the starting soil substrate containing leachable copper and other metals such as aluminum (Appendices A and B) that these elements are leaching from the substrate into the water, putting additional treatment demands on the systems. It is recommended that for construction of the full-scale systems, soils with low total and leachable copper concentrations should be used.

Table 5. Total and Leachable Soil Copper Concentrations in First Year of Operations.

Cell	Sampling Date	Days in Operation	Sample Depth (cm)	Total Cu (mg/kg)	SPLP Cu (mg/L)
CWTS 1A	28-Aug-14	1	Initial Composite	-	0.298 (0.148-0.608)
	19-Jun-15	58	0-5	960	0.055[1]
	16-Aug-15	116	10-20	950	0.187
	18-Sep-15	149	10-20	1300	0.049
	29-Sep-15	160	10-20	910	0.069
CWTS 1B	28-Aug-14	1	Initial Composite	-	0.298 (0.148-0.608)
	19-Jun-15	58	0-5	1400	0.033[1]
	16-Aug-15	116	10-20	1400	0.209
	18-Sep-15	149	10-20	830	0.065
	29-Sep-15	160	10-20	880	0.059
CWTS 2A	28-Aug-14	1	Initial Composite	-	0.298 (0.148-0.608)
	19-Jun-15	58	0-5	1175	0.037[1]
	16-Aug-15	116	10-20	660	0.139
	18-Sep-15	149	10-20	880	0.081
	29-Sep-15	160	10-20	1000	0.073
CWTS 2B	28-Aug-14	1	Initial Composite	-	0.298 (0.148-0.608)
	19-Jun-15	58	0-5	1100	0.039[1]
	16-Aug-15	116	10-20	1000	0.201
	18-Sep-15	149	10-20	830	0.078
	29-Sep-15	160	10-20	540	0.059

[1] Samples collected in June 2015 were at a shallow depth (0-5 cm) and copper content had therefore likely already been removed by washing from the faster flows of the CWTS system. The aging of the deeper (10-20 cm) soil copper concentrations over time to a less soluble form is shown in Figure 4.

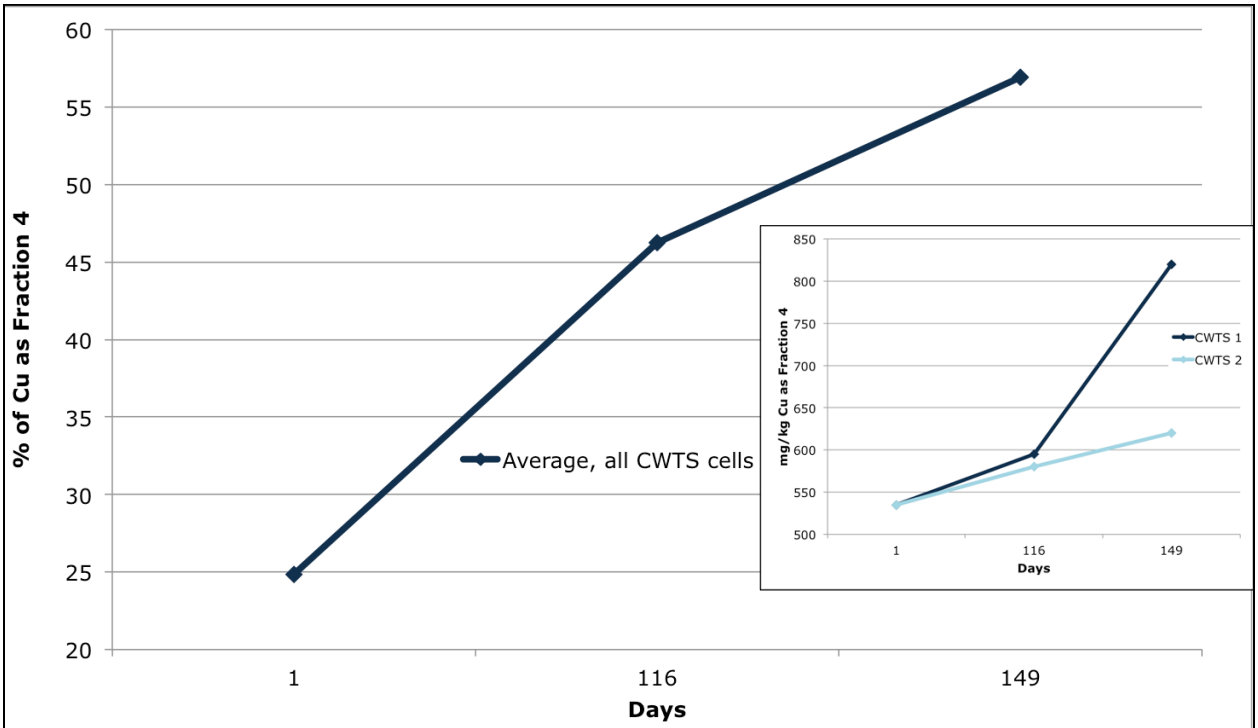


Figure 4 – Copper as sequential leach fraction 4 over time.
Copper in the form of the targeted Fraction 4 (sulphide mineral form and bound to organics), increased in the CWTS over time as the soils matured. Additional information in Appendix B.

6.2.Water

Pilot-scale testing was performed with water that mimicked the worst-case long-term post-closure water chemistry (mean influent concentrations of 146 µg/L copper and 10.2 µg/L selenium). In contrast, the on-site demonstration-scale system uses water from the W36 pond and therefore has inflow concentrations that are currently occurring during operations (mean influent concentrations of 58 µg/L copper and 6 µg/L selenium). Because of this, a lower percent removal will achieve the same extent of removal and associated outflow concentrations than was required in the pilot-scale systems. For example, the pilot-scale system achieved on average 92% removal of copper (mean outflow of 11.3 µg/L) and 41% removal of selenium (mean outflow of 6 µg/L). To achieve these same outflow concentrations with the demonstration-scale system, 80% removal of copper and minimal (or no) removal of selenium would be needed.

The treatment of copper has been improving through commissioning as the system has matured (Figure 7). However, as shown in Figure 5 and Appendix B, several elements are being released through the wetland by the soils, such as aluminum, cadmium, and copper (discussed further in Section 6.1). Because of this, negative percent removals of copper have been observed, especially in early operations (Figure 7). It was initially theorized that this was occurring when the May and June 2015 data reported outflow concentrations higher than the influent. This paired with the soils data (Section 6.1 and Appendix A) led to the design of a new sampling scheme to determine whether detectable levels of metals were being released from the soils within the wetland. For two timepoints (August 15, 2015 and

September 17, 2015), water was sampled not only at the feed and the outflow of each cell, but also through the cell. Beginning at the outflow of cell B and working upwards towards the feed, samples were taken at the B cell outflow, 1 m upstream of the B cell outflow, B cell mid point, 1 m downstream of the B cell inflow, and then the same 4 points for the A cell (Figure 6). In each case, the sample was taken from within a reaching distance of the side shore to ensure that sediments were not suspended in sampling.

It was found that significant metals concentrations were leaching from the wetlands soils of all cells (Figure 5). Moreover, metal types and concentrations (e.g., Aluminum, Appendix B) that could not be accounted for by the influent water chemistry were elevated at random points within the wetland. This suggested that because of metals leaching from the soils, the treatment occurring within the wetlands was far greater than what was being observed by simply measuring the inflow and outflow points (Figure 7). For example, in August of 2015, influent copper concentrations were measured as 50 µg/L, but within the first meter of cell 1A increased to 100 µg/L, and by the outflow of cell A, down to 42 µg/L (58 µg/L decrease), they then increased to 70 µg/L at the beginning of cell B, and again were treated to a final outflow concentration was 40 µg/L (30 µg/L) for a total of at least 88 µg/L removed by Series 1 of the treatment wetland. This suggests the wetland is actually achieving much greater copper treatment (88 µg/L removed), than would be suggested by only measuring the inflow and outflow of the system (suggests 10 µg/L removed). Similar leaching was observed through the wetland during testing in September 2015.

Copper, cadmium, zinc, and other metals and metalloids are being noticeably decreased in concentration from the water entering the wetlands (Figure 7 and Appendix B), while other metals/metalloids of potential concern are below background water quality for the area or below the detection limit in the influent of the wetland. Dissolved zinc had on average 65% removal in the wetland systems and by September 2015, the outflow concentrations were on average 20% and 50% lower than influent concentrations for dissolved copper and cadmium, respectively. This suggests that not only is the wetland maturing as expected, creating reducing sulphide-producing conditions, but that it is already performing far beyond what was anticipated from the design, sequestering orders of magnitude more copper than would be apparent by looking at the influent water alone (i.e., is also removing copper that has leached from the soils over time).

Removal rate coefficients (k) based on the HRT of each system would normally be calculated to compare treatment efficiency. However, because of the leaching of various elements from the soil, calculation of k for constituents of concern is not representative of what is ongoing in the system, as there is concurrent leaching and treatment occurring throughout a given cell (Figure 5). Furthermore, because the soil used for construction of the demonstration-scale system is different than what was used for pilot-scale systems, the hydrology and pore volume of the system is not known; a parameter that is needed for accurate calculation of the HRT and therefore k. As the same wetland substrate was used to construct both demonstration-scale systems, the removal rate coefficients can be estimated and compared between these two systems; however, comparison with removal rates achieved during pilot-scale testing is not possible until the HRT is defined. It is therefore recommended that a tracer study be performed in 2016 to assess the hydrology and pore volume of the CWTS (see Section 8).

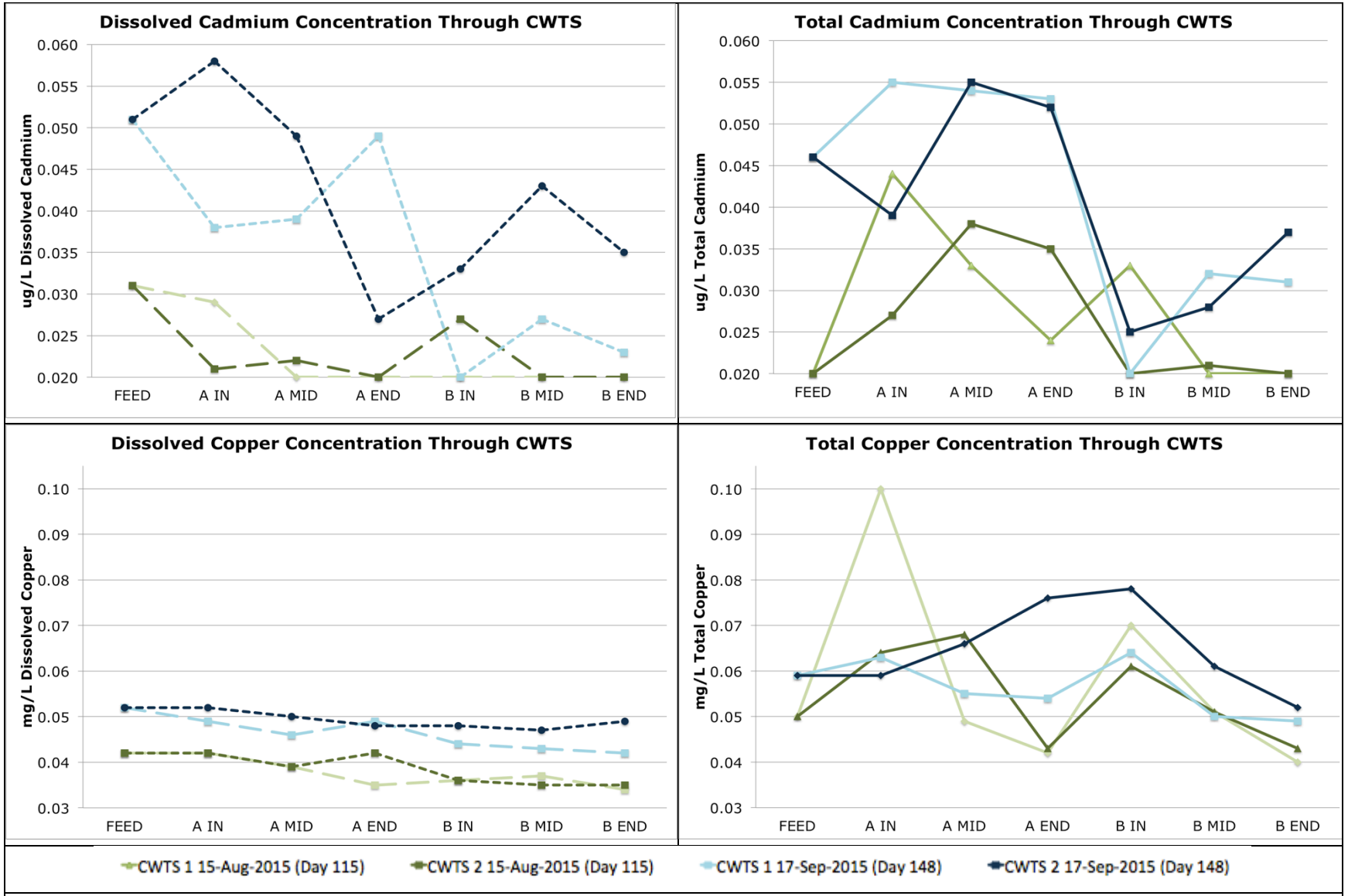


Figure 5 – Concentrations of key elements through the CWTS.
The y-axis values start at the detection limits.
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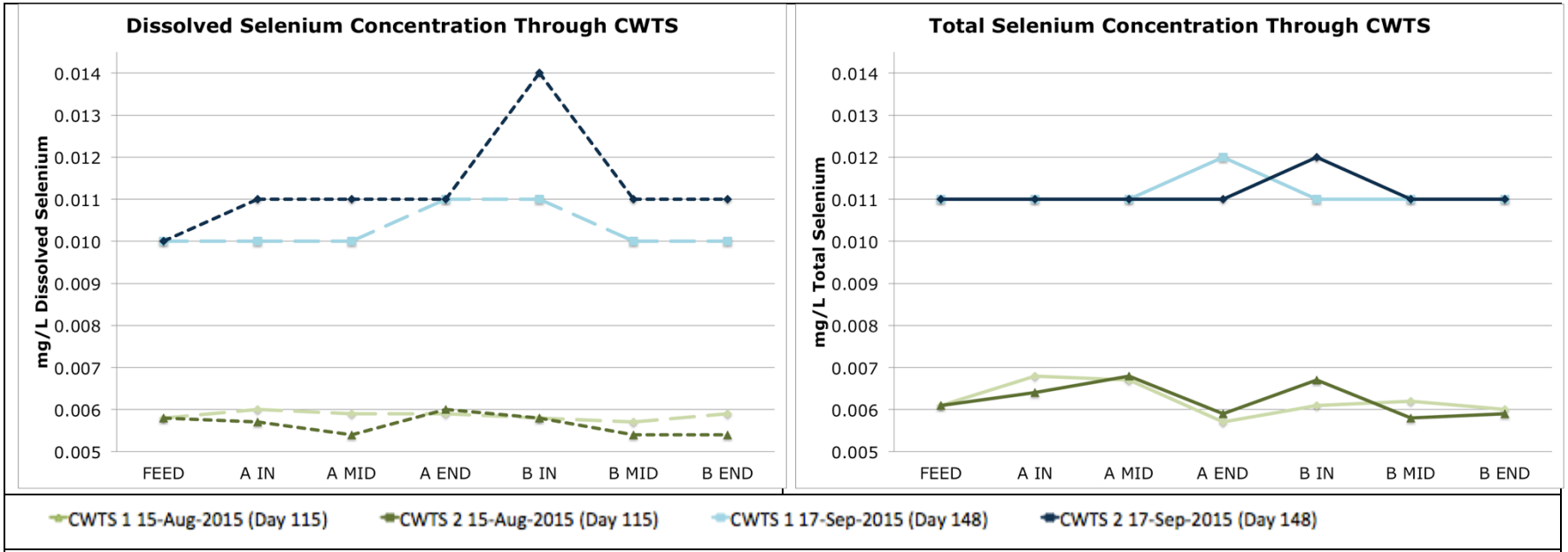


Figure 5 – Continued

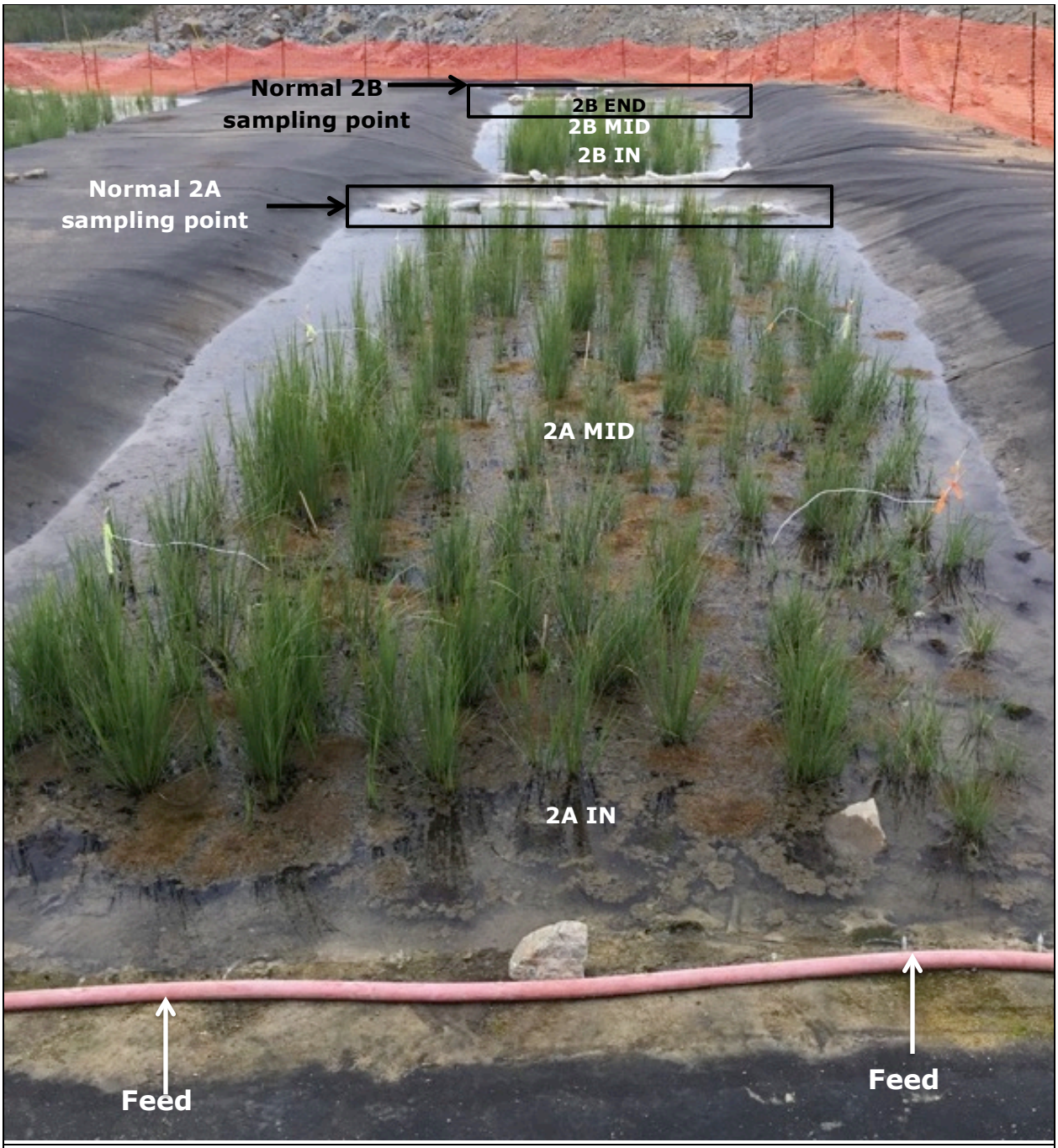


Figure 6 – Sampling points through the CWTS.

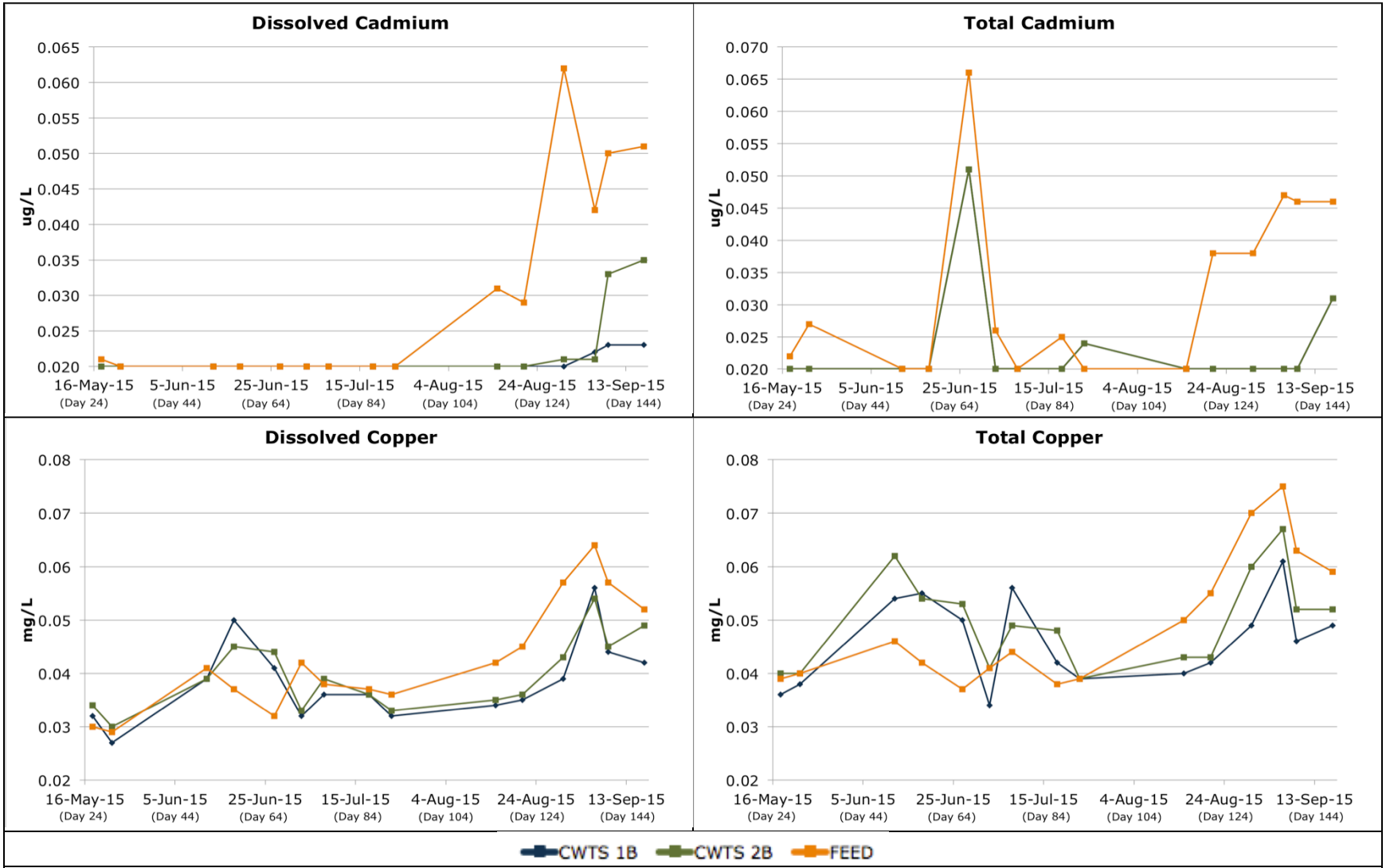


Figure 7 – Element concentrations in the demonstration scale wetland from May 17, 2015 to Sept 17, 2015.
The Measurement Uncertainty (MU) for cadmium is +/- 0.020 µg/L. The MU for copper and selenium is +/- 0.0002 mg/L. Timepoints where the dissolved cadmium concentration is higher than the total concentration are within the limit of uncertainty (Maxxam Analytics). Detection limits for cadmium, copper, and selenium are at or below y-axis values.
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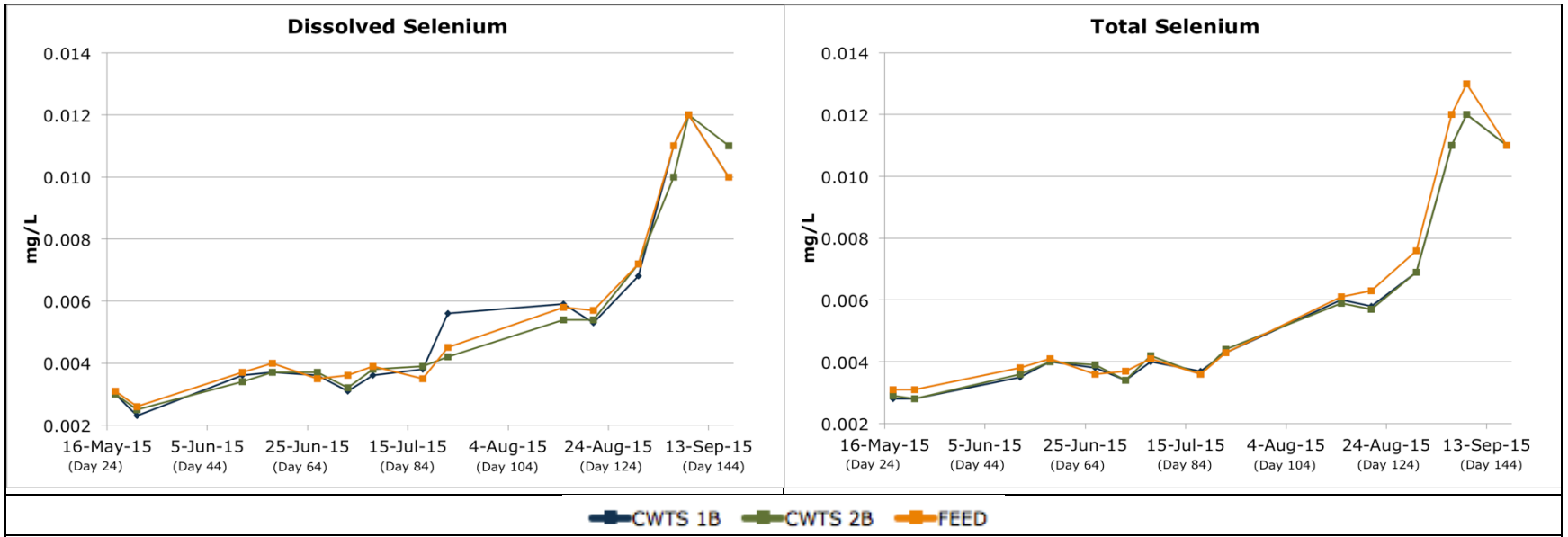


Figure 7 - Continued

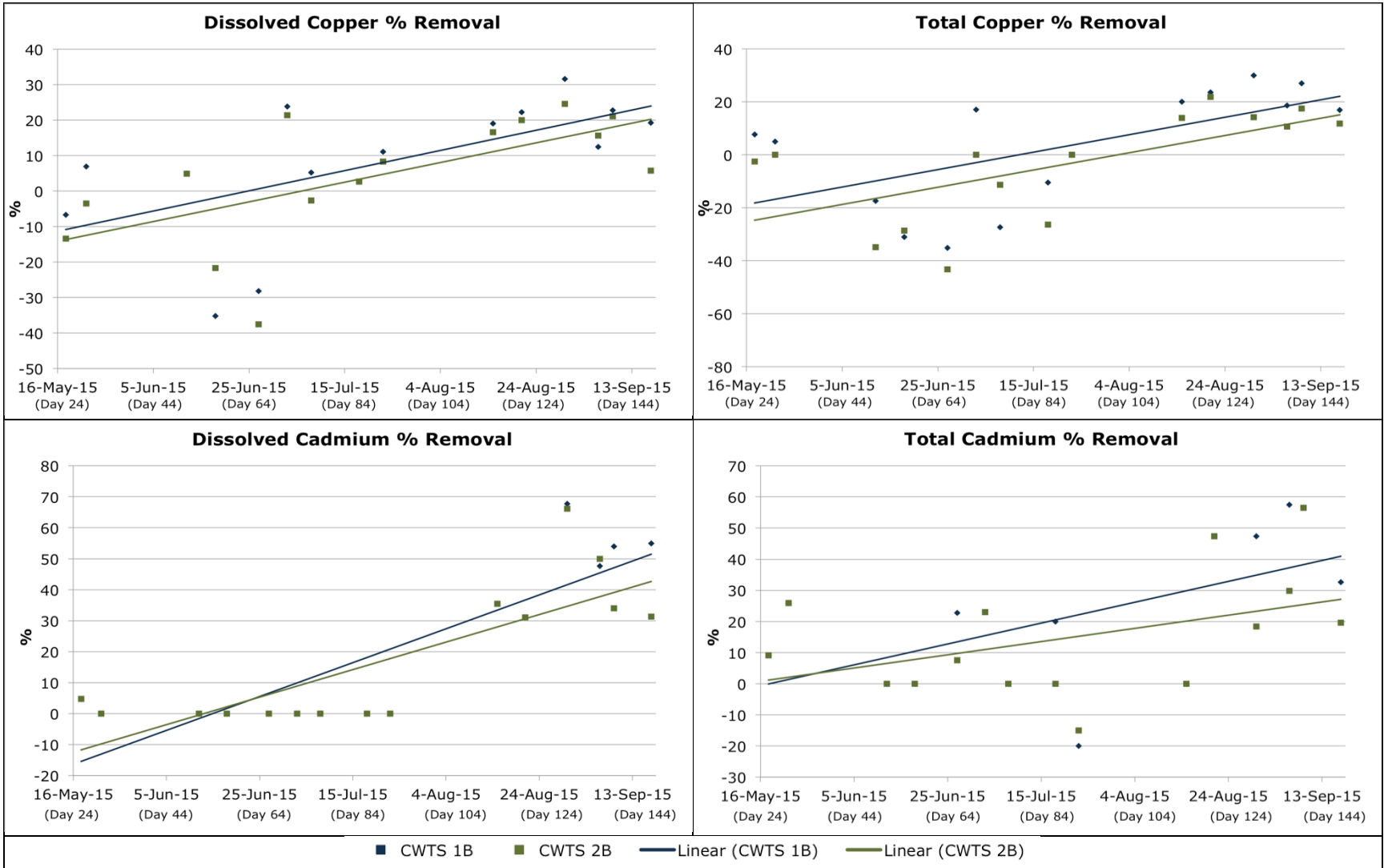


Figure 8 – Percent removal of elements in the demonstration scale wetland from May 17, 2015 to Sept 17, 2015.
If the concentration was below the detection limit of the assay, the detection limit was used to calculate the percent removal. Note the Measurement Uncertainty (MU) for cadmium was +/- 0.020 µg/L. The MU for copper and selenium is +/- 0.0002 mg/L (Maxxam Analytics) (Appendix A). The percent removal calculations for cadmium are therefore estimations within this range of error.
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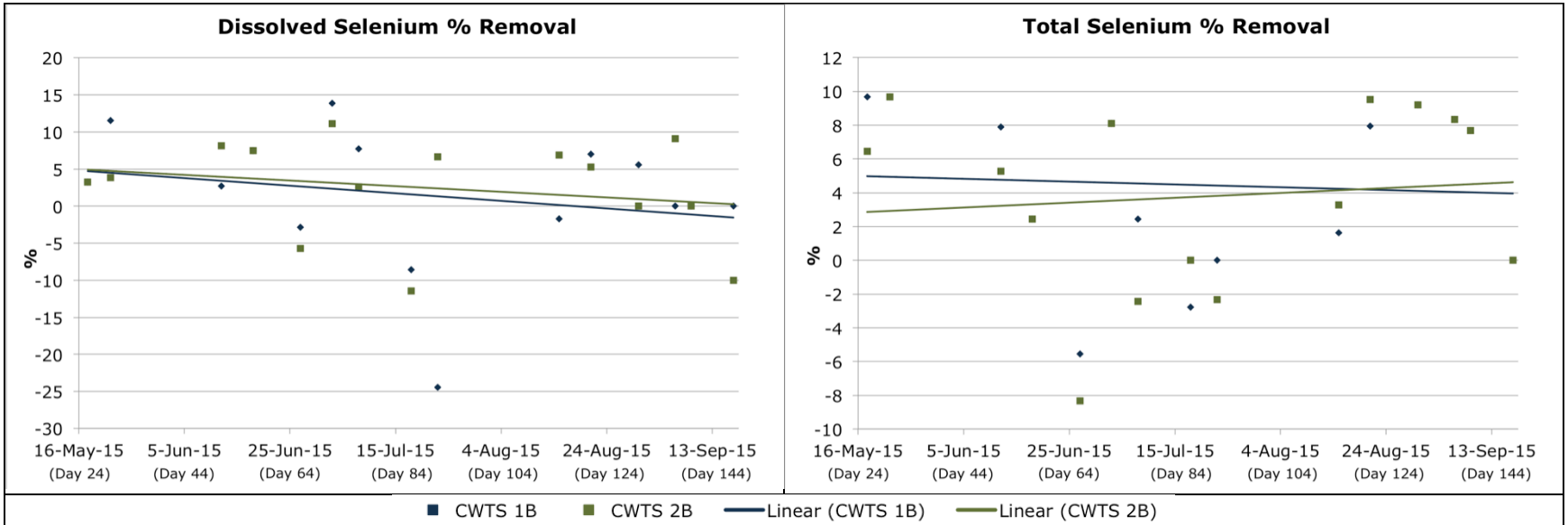


Figure 8 – Continued

6.3.Chelated iron (fertilizer micronutrient) treatment

Because elevated metals were present in the soils used for construction, there was an opportunity to assess whether non-invasive treatments could be used to further leach the metals and improve soil quality. A chelated iron (Fe-EDTA) product was added to the Series 2 wetland on September 17, 2015. The product selected is Fe-EDTA called Plant-Prod 13.2% Iron Chelate (containing 68% of the chelating agent Ethylenediaminetetraacetic acid (EDTA)), and is routinely used in the agricultural industry as a micronutrient supplement for iron-deficient soils. Fe-EDTA is also commonly used in remediation of metals contaminated soils (e.g., Thayalakumaran *et al.,* 2003).

The Fe-EDTA dissociates based on the pH of the water, and the free EDTA will be available to bind other elements such as copper and remove them from the soils, leaving the Fe in the wetland. Fe-EDTA was chosen instead of fertilizer with EDTA (which is also a commonly used agricultural product), as the fertilizer would provide additional nitrogen sources, which can interfere with selenium treatment, and phosphate which could cause an algal bloom that would be detrimental to the active water treatment facility at Minto that receives the CWTS outflow water. The Fe-EDTA treatment was performed on Series 2, with Series 1 remaining untreated for comparison purposes.

After addition of Fe-EDTA, copper concentrations in the wetland water for both CWTS Series 1 and 2 were monitored daily by Minto's internal laboratory. Increases in copper concentrations were observed in the outflow from both cells in CWTS Series 2 after the addition of Fe-EDTA (Figure 9). Unfortunately, the CWTS feed copper concentration also increased, as the outflow from the demonstration-scale wetlands reports to the same area as the feed water draws from at the W36 area. As such, the leached copper (and presumably, also EDTA) is recirculating through the wetland (including Series 1).

In the 10 days following the addition of Fe-EDTA, approximately 75 g of copper was removed from the wetland Series 2. This is approximately 1.25 g of copper per m^2 of wetland. Testing of the soils at different depths in Spring 2016 will determine how much copper has been removed from the wetland as a result of this treatment.

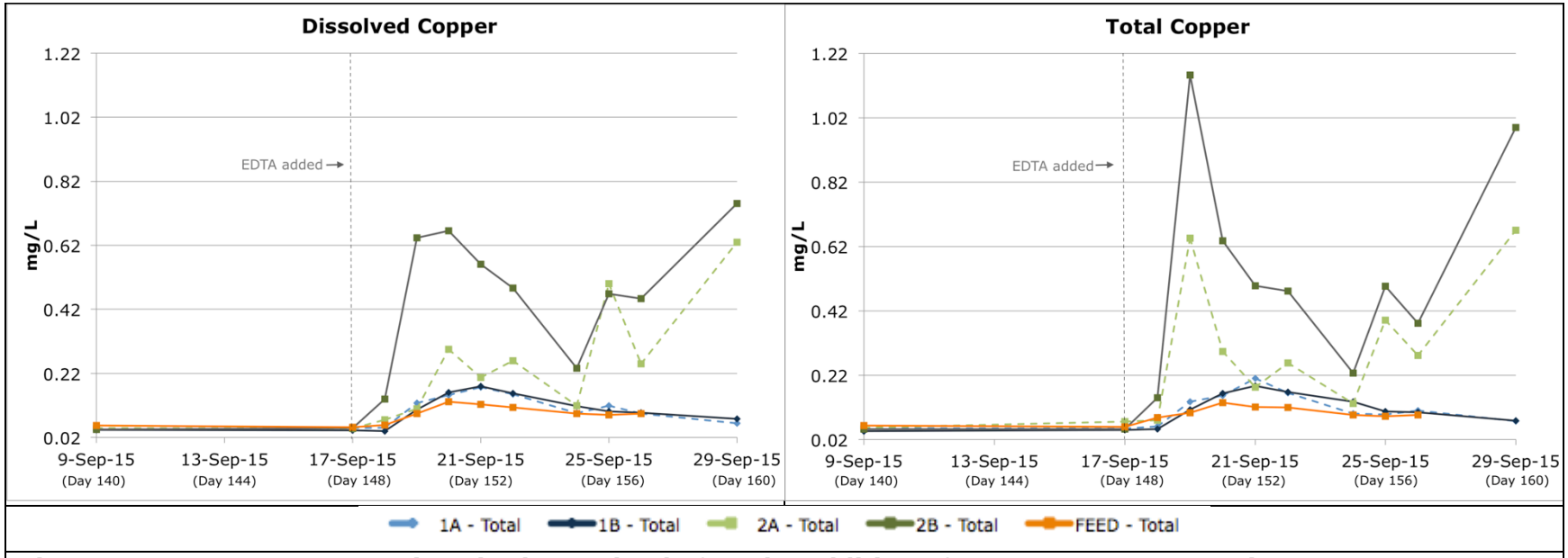


Figure 9 – Copper concentrations in the wetland after the addition of Fe-EDTA to CWTS Series 2
Detection limits are below y-axis values. No influent (feed) was tested September 29th as the system had frozen and was turned off for the winter.

6.4. Plants

During the commissioning period, plants establish and mature, with density expected to increase over time. When counted at the first site visit in June 2015, >95% of the plants were found to have survived their first winter. This survival rate is impressive considering the late fall planting of August 28, 2014 and suggests the *C. aquatilis* are very robust and reaffirms they are a good candidate for use in the full-scale CWTS at Minto. Additionally, there was a greater than 20% increase in total stems in the short growing season up to June 2015 (as counted by new shoots/runners). By the second site visit in August 2015, the plants were too dense to count stems, and therefore the survival and establishment was considered a success and no longer monitored by counting (Figure 6). Despite several mosses drying out during the winter after planting, there was 100% survival and establishment. However, their growth was much slower than the sedge and therefore they were supplemented with additional aquatic moss (harvested from the same W10 area as the original plants for the CWTS) in June and August 2015 to augment their general abundance and bring the CWTS through commissioning more quickly (Figure 10).

Pilot-scale testing suggested that *C. aquatilis* would increase more quickly the number of stems once the maximum plant height was achieved, and also that maximum plant height is dependent upon water depth (with deeper water resulting in taller plants). This was also seen in the demonstration-scale wetland on site, with the density of plants filling in greatly through July and August (Figure 11). It is anticipated that the *C. aquatilis* will have grown uniformly across the wetland by end of the 2016 or 2017 growing season.

Of note, it was observed that during the last site visit of the year by Contango (September 17-18th, 2015), the *C. aquatilis* in the demonstration-scale CWTS were still predominantly green, while other plants in the area had turned brown. This suggests that the CWTSs may have a longer range of activity than natural systems in the area, possibly due to liners and separation from colder ground water.

Concentrations of various elements in *C. aquatilis* and moss from the demonstration-scale systems were compared to those from pilot-scale testing. It is expected that the concentrations would be higher in the demonstration-scale system as the CWTS is not yet fully functional (i.e., elements are in greater bioavailability when in dissolved form) and also due to the composition of the soils used for construction. Accordingly, *C. aquatilis* and moss in the demonstration-scale system had higher copper concentrations than the pilot systems. However, concentration of copper in the *C. aquatilis* at the end of 2015 was not significantly different than that of the plants borrowed from the W10 area for planting of the systems in 2014 (Figure 12). The moss had higher copper and cadmium concentrations which is expected as it removes these elements in the wetland through sorption (and will eventually mineralize in subsequent years). As mosses are not a significant food source for any fauna, this is a safe mechanism to transfer elements from water eventually into stable mineral forms.

Cadmium concentrations were either below or close to the detection limit in *Carex* (Figure 12). Similarly, cadmium concentrations in moss and selenium concentrations in both plant types tested were lower in the demonstration-scale compared to pilot-scale. This result was presumably due to lower concentrations of both cadmium and selenium in the water of the demonstration-scale system than pilot scale. As mosses continue to grow and fill in the wetland, selenium treatment performance is expected to increase as mosses are a source of sorption, and also harbor the highest abundance of selenate-reducing microorganisms of all samples tested (see Section 6.5.2).



Figure 10 – Moss maturation in CWTS.

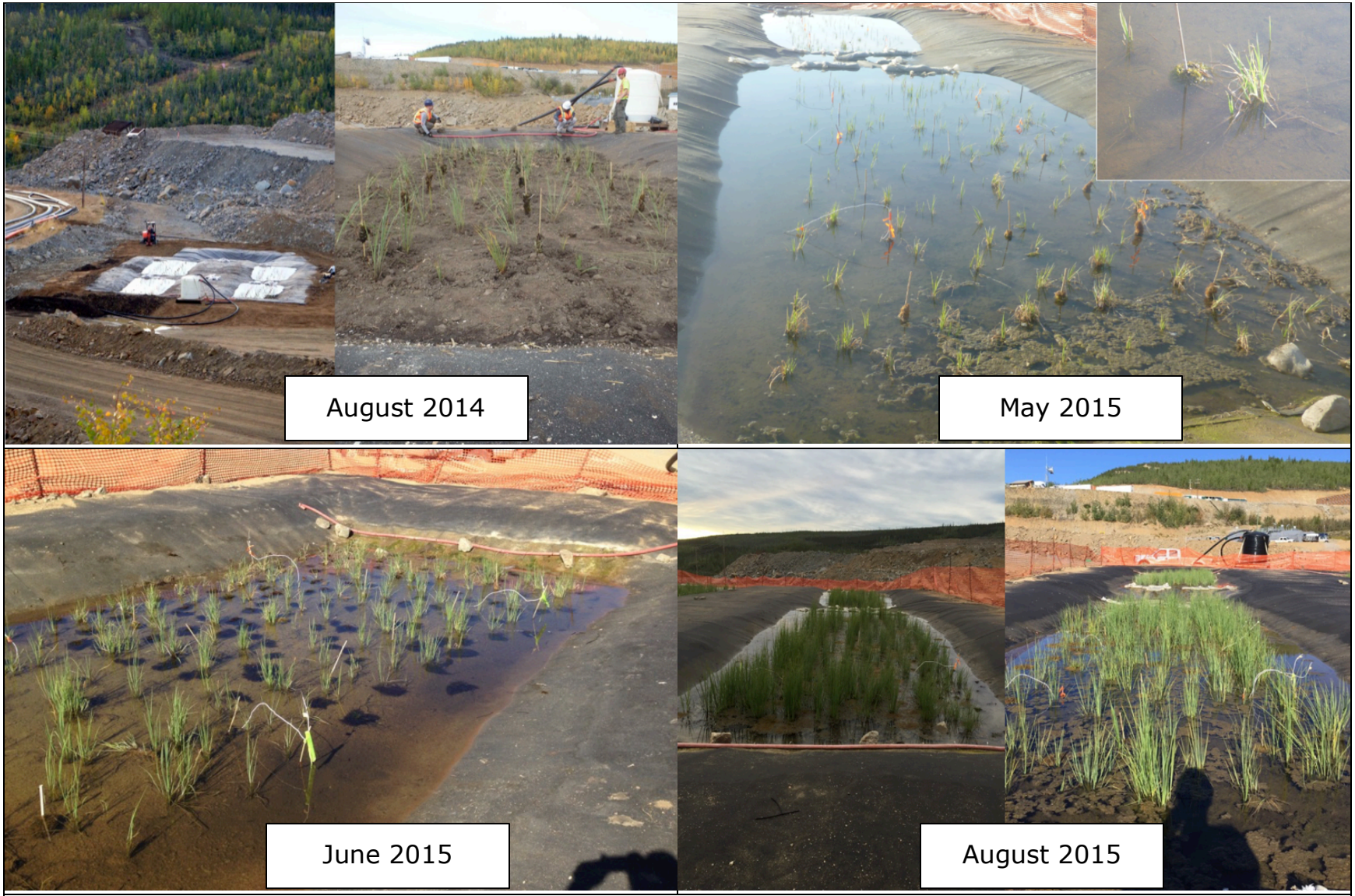


Figure 11 – Maturation of the CWTS from construction to year-end 2015.
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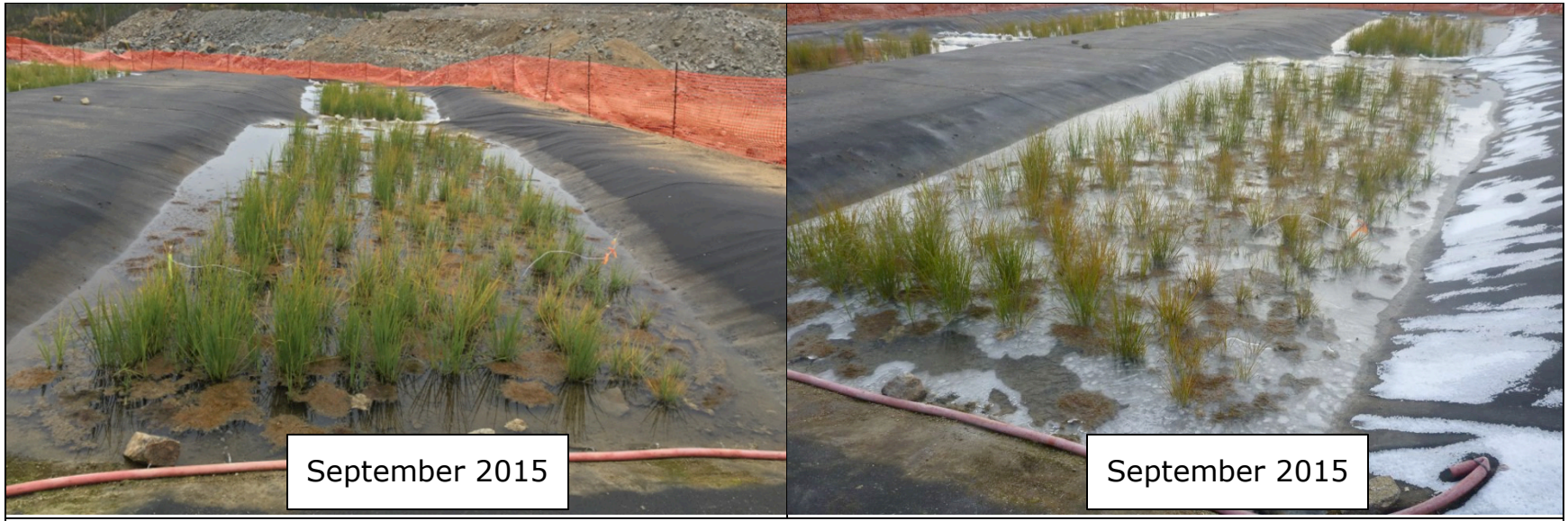


Figure 11 - Continued

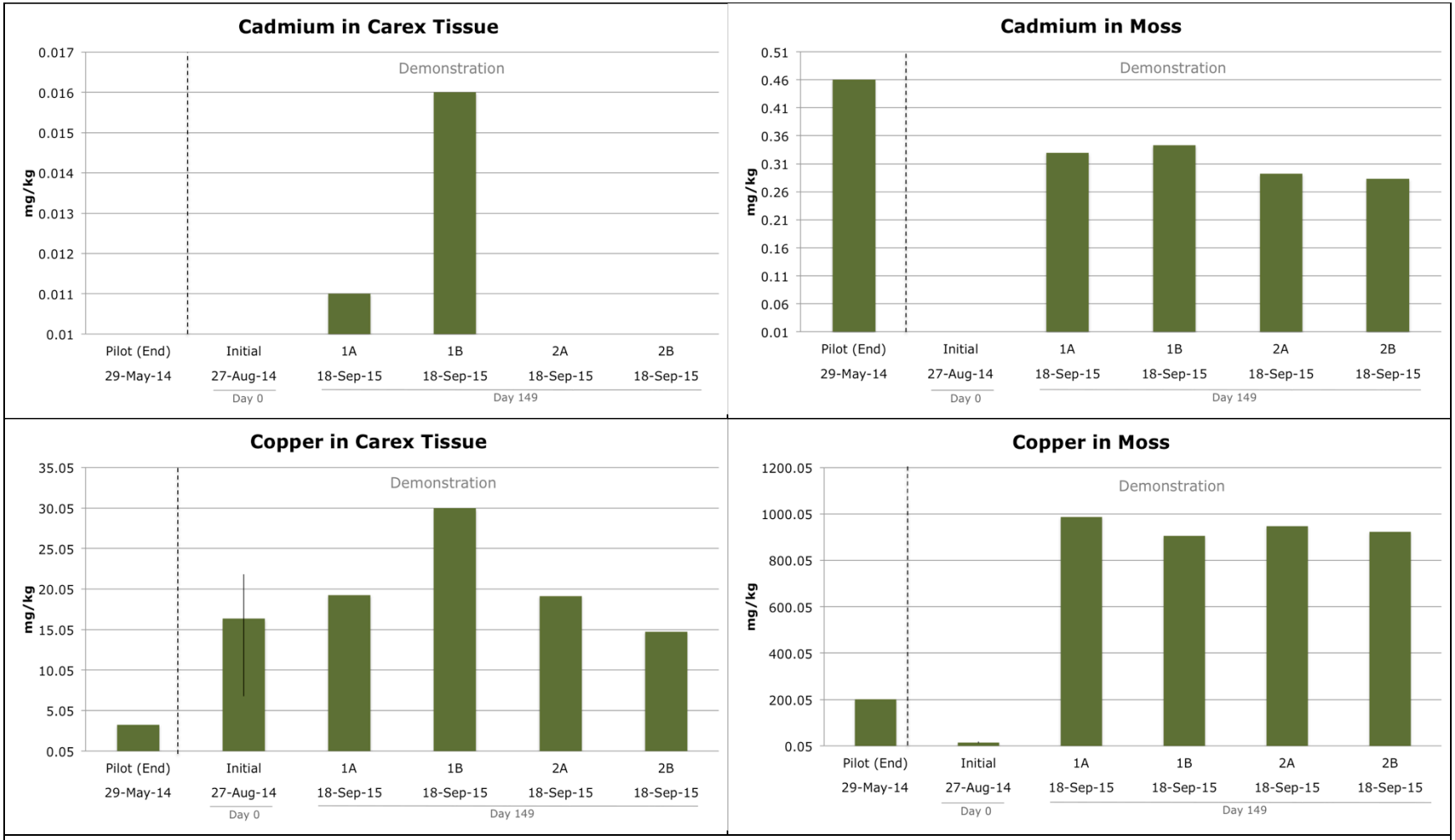
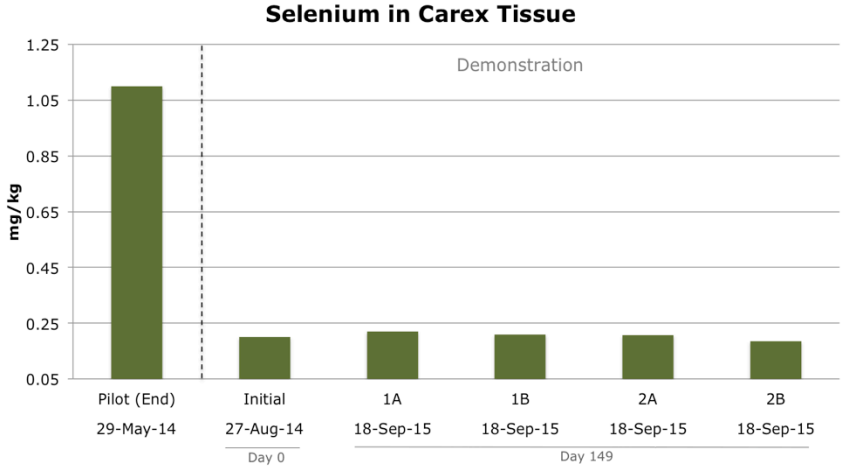


Figure 12 – Element concentrations in plant tissue.
Y-axis starts at the 2015 detection limit (DL). The 2014 DL for Cadmium is the same as 2015. The 2014 DL for Copper is 0.5 mg/kg, and Selenium is 0.2 mg/kg. The initial data set is the average value of CWTS 1A-2B. The error bars indicate the minimum and maximum values in that data set.
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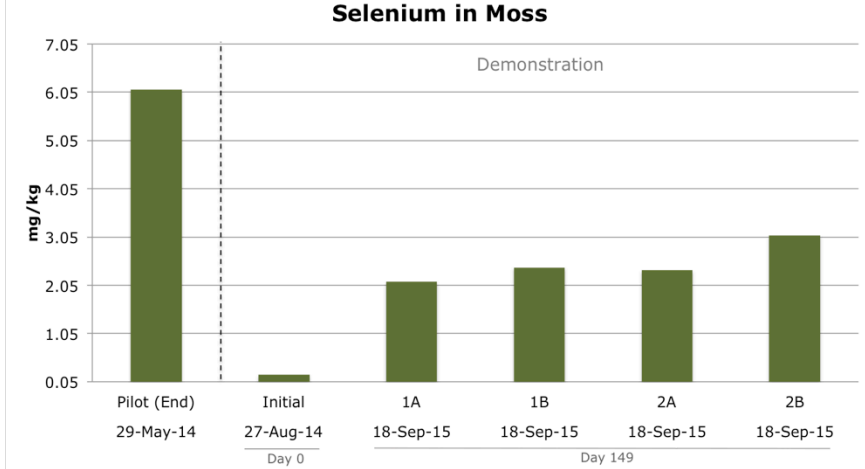


Figure 12 - Continued

6.5.Beneficial Microbes

Microbes are the driving force of many treatment pathways that are targeted in CWTSs. The beneficial microbes in these systems catalyze biogeochemical cycles for remediation of specific constituents of concern. Careful design of CWTSs can mimic the environmental conditions needed to enhance the abundance and metabolic activity of these beneficial microbes. Accordingly, complimentary methods of genetic and growth-based testing were used to characterize the microbial populations associated with a range of microbial habitats in the demonstration-scale CWTS (e.g., soils, sediment, biofilms, aquatic mosses, and plant roots).

In the context of the Minto Mine CWTSs, beneficial microbes include those that are involved in the reduction of selenium (selenate and selenite), nitrate, and sulphur compounds (which in turn treats copper and cadmium). Information on each of these mechanisms and the associated microbial populations in the demonstration-scale system is outlined in the following sections.

6.5.1. Sulphide-producing bacteria

Treatment is achieved by targeting the lithic biogeochemical sequestration of divalent metals through sulfide (i.e., S^{2-}, HS^{-}) precipitation as mineralized species (e.g., chalcocite [CuS], covellite [Cu_2S]). These sulfide bound species are relatively insoluble (CuS; ksp=10^{-16}; Stumm and Morgan 1996), and are transferred from the water column into the CWTS soil as non-bioavailable fractions (Murray-Gulde et al., 2003; Huddleston et al., 2008). Moreover, similar reactions occur with cadmium rending it non-bioavailable. As such, sulphide production is a key biogeochemical treatment mechanism for water treatment at Minto Mine. Sulphides can be created by beneficial microorganisms through the reduction of sulphur-containing compounds, such as sulphate, sulphite, thiosulphate, and elemental sulphur.

Based on the information gathered in pilot-scale testing, the targeted soil redox for the demonstration-scale CWTS is between -100 and -250 mV. This is in agreement with literature that indicates anaerobic conditions with relatively low ORP (-250 to -100 mV) are necessary for promoting anaerobic metabolisms in bacteria which oxidize organic matter, producing electrons which reduce sulfate to hydrogen sulfide (H_2S) and other reduced sulfide species (i.e. bisulfide ion (HS^{-}), sulfide ion [S^{2-}]; Mitsch and Gosselink 2007). In these redox ranges, bacterial sulphide-production through reduction of sulphur compounds is expected, alongside increases in the proportion (percentage) and abundance of these microbes.

The percentage of sulphate-reducing bacteria in the community increased over time in the demonstration-scale CWTS (Figure 13), as would be expected during the commissioning period. The percentage of sulphate-reducing bacteria is similar to what was present in pilot-scale systems at a comparable time point in operation (i.e., day 114 vs. day 80 in Figure 13), however, the types of microbes differ. Conversely, when considering all sulphide-

producing bacteria (i.e., including those capable of sulphur, sulphate, sulphite, and thiosulphate reduction), the demonstration-scale systems have a higher percentage than the pilot-scale systems at a comparable time point (Figure 14), indicating the commissioning period is proceeding well. Sulphide-producing organisms identified during the site assessment at W10 and in low proportions at W15 (Contango, March 2014) were also present in both the pilot- and demonstration-scale systems, indicating that harvested plants were also a source of beneficial microbes.

As expected, the proportion of sulphide-producing bacteria increased as the soil redox decreased (Figure 14). Furthermore, the overall abundance of sulphide-producing bacteria has increased in the soil of the CWTSs over time as reducing conditions are achieved during the commissioning period (Figure 15). This is particularly notable in Series 1, which achieved targeted soil redox conditions quicker than Series 2, which is still establishing reducing conditions (see Section 5). The soils and *C. aquatilis* roots had the highest abundance of sulphide-producing organisms of all sample types tested, with the deeper soil (10-20 cm) having higher abundances than the shallow soils (0-5 cm) at day 149. As expected, algae and biofilm samples had very low abundances of sulphide-producing bacteria. Interestingly, the aquatic mosses had a large variability in the abundance of sulphide-producing bacteria (sometimes having as many or more than the soils). This variability may be related to general growth habits of the moss, with live, photosynthetic (oxygenating) parts of the moss having fewer sulphide-producing bacteria, while the older, decomposing regions have more. This fits with the targeted processes in the CWTS, where metals are first sorbed to moss, then eventually mineralized through sulphide-production.

The microbial analysis of various sample types in the CWTS has therefore confirmed that the commissioning period is proceeding as expected, with beneficial sulphide-producing bacterial populations establishing in the soils alongside reducing conditions.

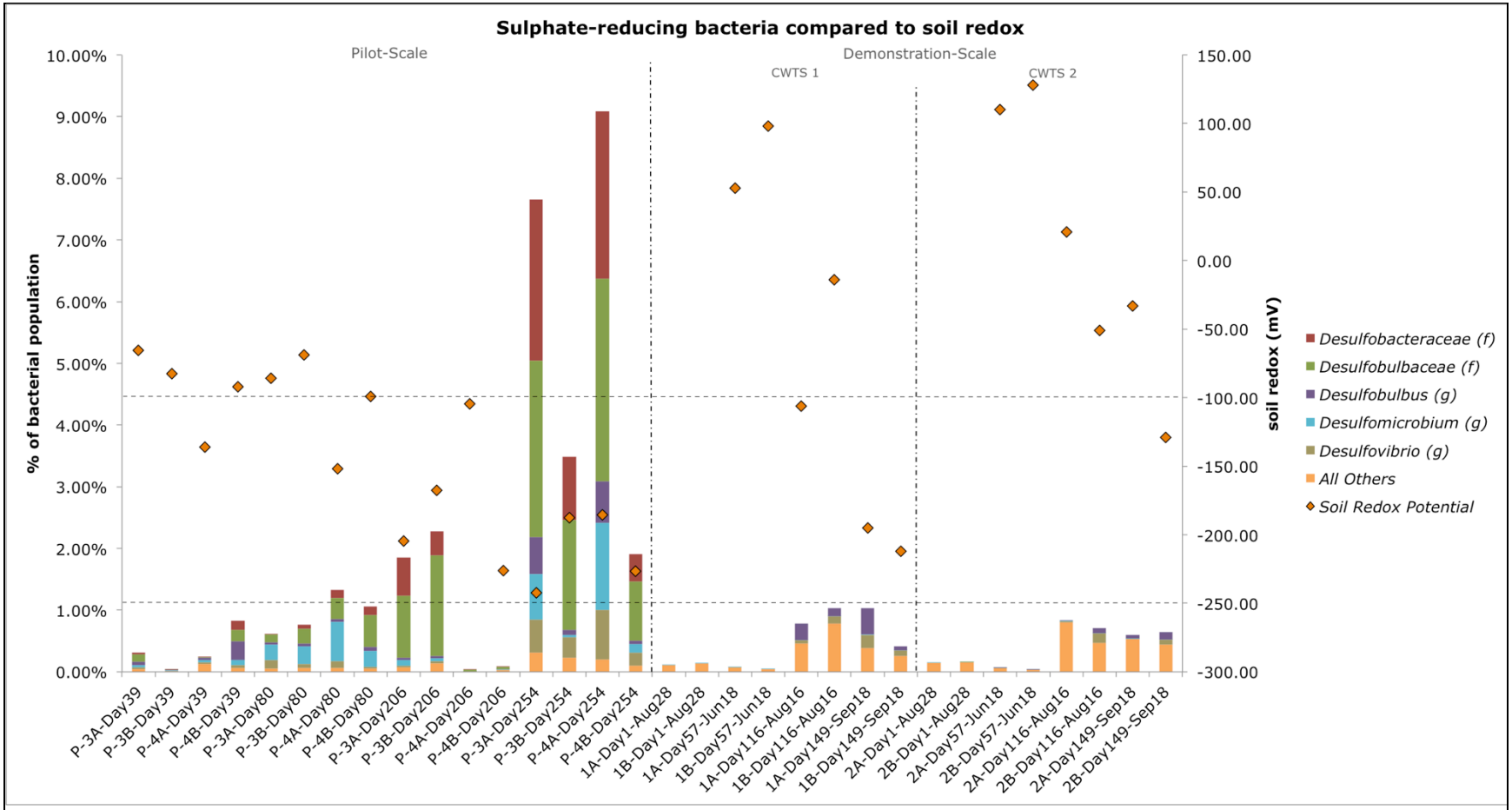


Figure 13 – Percentage and identity of sulphate-reducing bacteria in soil of the pilot-scale and demonstration-scale CWTSs

The vertical dashed lines separate the pilot-scale system data from the demonstration-scale system 1 and system 2 data. The horizontal dashed lines indicate the targeted soil redox range.

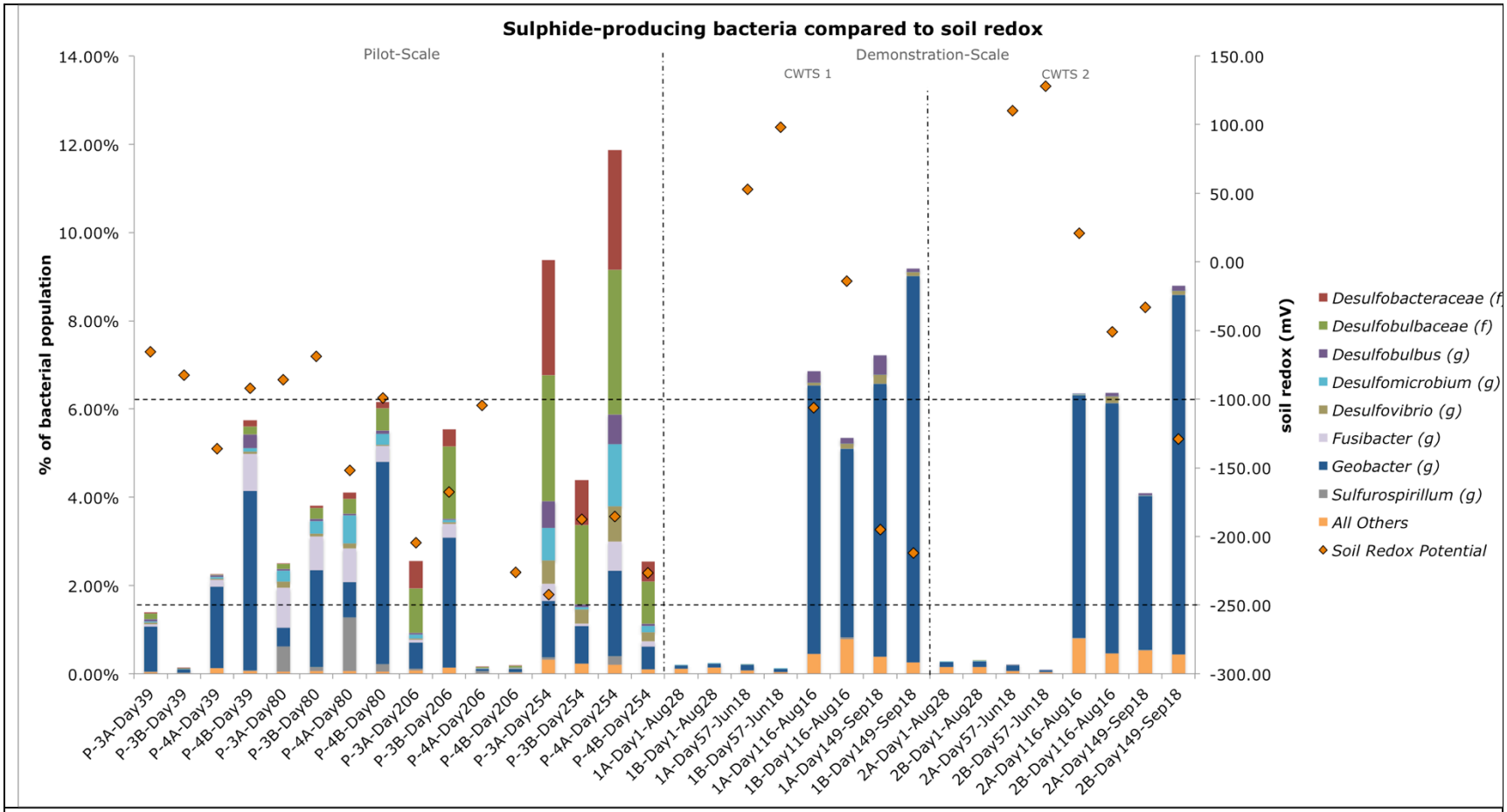


Figure 14 – Percentage and identity of sulphide-producing bacteria in soil of the pilot-scale and demonstration-scale CWTSs

The vertical dashed lines separate the pilot-scale system data from the demonstration-scale system 1 and system 2 data. The horizontal dashed lines indicate the targeted soil redox range.

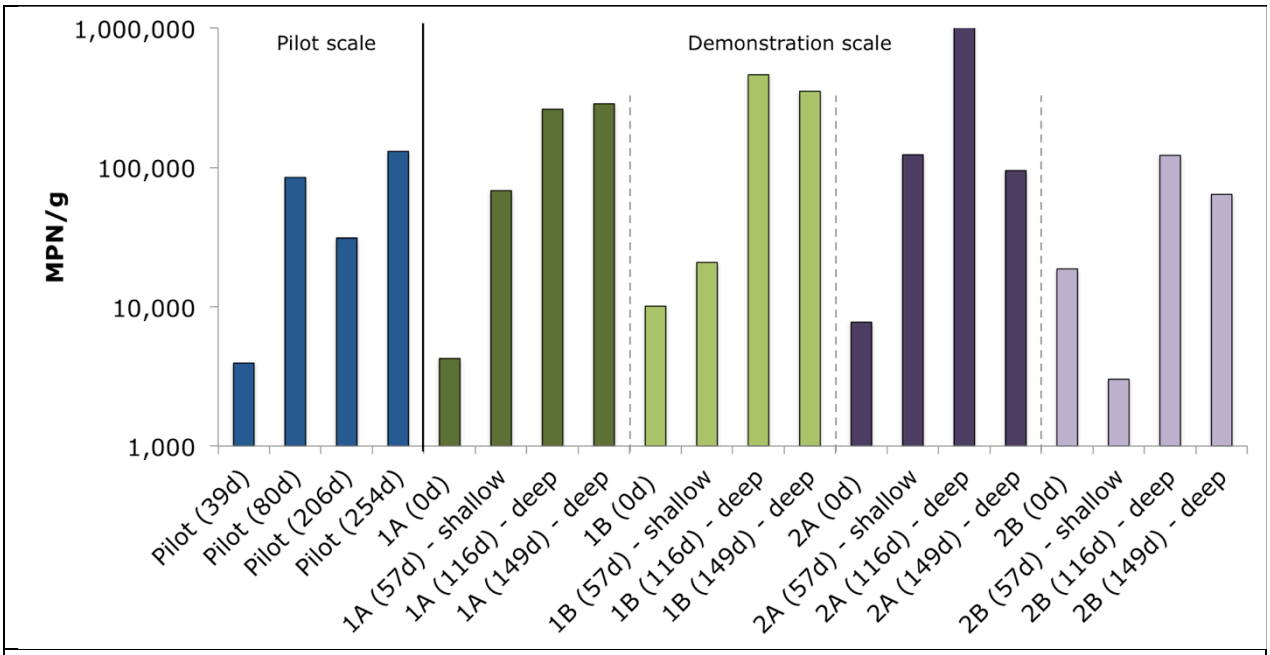


Figure 15 – Inferred abundance of sulphide-producing bacteria in soil of the pilot- and demonstration-scale CWTSs
Inferred abundance is calculated based on a combination of genetic and growth-based data.
- shallow soil (0-5 cm); deep soil (10-20 cm); soil at 0d was that used for construction of wetland

6.5.2. Selenium-reducing bacteria

The targeted selenium treatment pathways in the Minto CWTS include sorption to moss and soils, and subsequent microbial reduction of soluble (sorbed) selenate (Se(VI)) and selenite (Se(IV)) to insoluble elemental selenium (Se(0)). This reductive process can also be achieved directly in the water column, but is more effective associated with mosses and biofilms due to their sorptive properties that bring the selenium in contact with beneficial selenium-reducing bacteria. This is achieved within the range of soil redox conditions targeted for sulphate-reduction as suggested by pilot-scale testing and literature (see Section 5).

Selenite-reducing organisms are ubiquitous in nature and as expected, were detected in all sample types, including algae, biofilm, moss, soil, sediment, and roots. Although selenate-reducing organisms are generally less abundant in the environment, they were found associated with all sample types, indicating that the conditions conducive to their proliferation have been created within the CWTS. Moreover, the abundance of selenite- and selenate-reducing organisms increased over time in the demonstration-scale system during the commissioning period (Figure 16). Aquatic mosses were found to host the highest abundances of both selenate- and selenite- reducing organisms, affirming the importance of the inclusion of moss in the CWTS. Additionally, shallow soils (i.e., top few centimeters, including associated biofilms) were found to host an abundance of selenite-reducing organisms, particularly in Series 1. The abundance of selenite-reducing organisms associated with the demonstration-scale moss and shallow soils is similar if not higher than

that observed in the soil of the pilot-scale testing, indicating the commissioning period is proceeding as anticipated (selenate-reducing organisms were not tested for at pilot-scale).

6.5.3. Nitrate-reducing bacteria

Nitrate is sometimes a constituent of concern during operations and early closure owing to residuals from blasting activities. Even if not in exceedance of water quality guidelines in terms of receiving environment objectives, nitrate often requires attention in order to achieve treatment of other constituents. The presence of nitrate can interfere with the treatment of certain elements in water (such as selenium). Nitrate can be removed from water by different types of microbes, including nitrate reducing bacteria which are capable of reducing nitrate (NO_3) to nitrite (NO_2), and also denitrifying organisms that are capable of fully reducing nitrate to nitric oxide (NO), nitrous oxide (N_2O), and dinitrogen gas (N_2 - which is the most abundant gas in air). MPN analysis was therefore used to quantify these organisms.

Nitrate-reducing and denitrifying organisms were found associated with all sample types in the demonstration-scale CWTS (Figure 17). Soil samples had an increase in nitrate reducers over time, with the highest abundance of all samples found in the deeper soil sample from day 149. In contrast, denitrifying organisms were found to be associated primarily with the shallower soil depths and also a high abundance associated with moss, which increased in abundance over the commissioning period.

Although the pilot-scale systems had higher nitrate concentrations (30 mg/L as N during High Nitrogen testing phase) than the demonstration-scale systems (average 10 mg/L as N), the abundance of nitrate-reducing and denitrifying organisms are both similar. As such, these results also confirm that the demonstration-scale system is establishing as expected during the commissioning period.

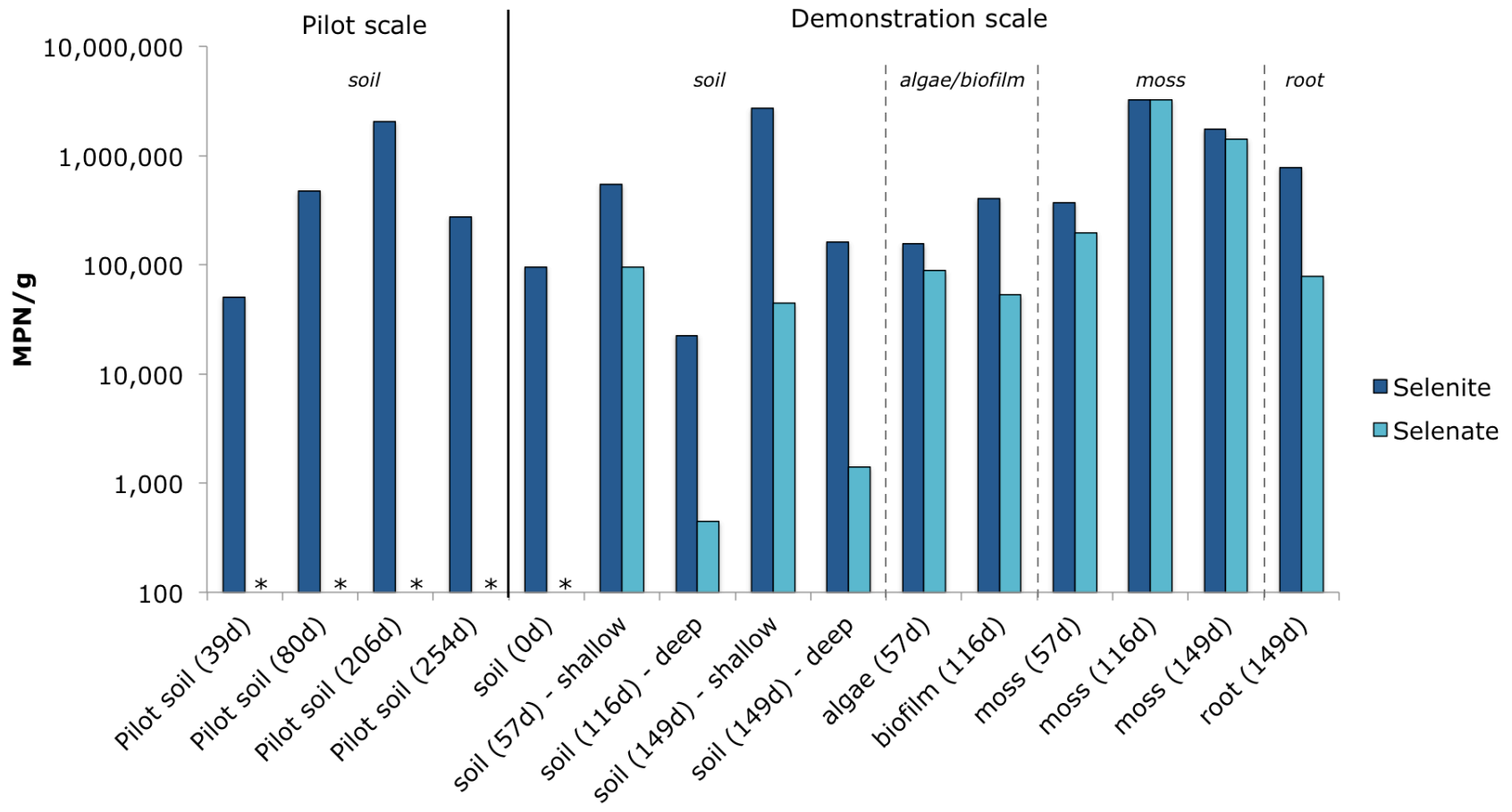


Figure 16 – Abundance of selenite- and selenate-reducing organisms in various CWTS sample types over time
* - pilot-scale system and initial timepoint for demonstration-scale system not tested for selenate-reduction
- shallow soil (0-5 cm); deep soil (10-20 cm); soil at 0d was that used for construction of wetland

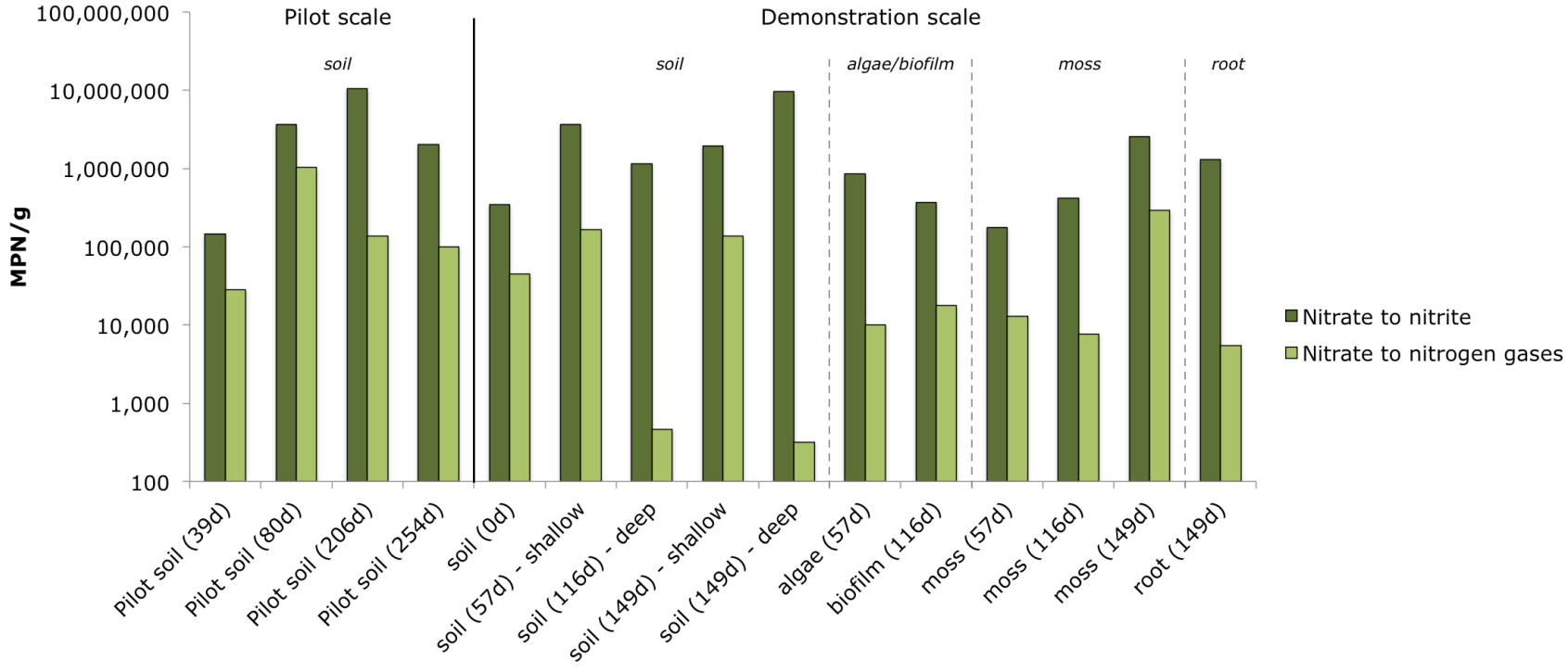


Figure 17 – Abundance of nitrate-reducing and denitrifying organisms in various CWTS sample types over time
Nitrate-reducing = reduction of nitrate to nitrite; denitrifying = reduction of nitrate to nitrogen gases. Shallow soil (0-5 cm); deep soil (10-20 cm); soil at 0d was that used for construction of wetland.

7. Summary of Results

Table 6 summarizes results and findings from 2015 commissioning of the demonstration-scale system at Minto.

Table 6. Summary of Minto Demonstration-scale 2015 Testing

Objective	Purpose	Key Findings
Evaluate construction	Optimize construction and effectiveness of operation of full-scale systems	**Layout** -Outflow collection pond should have outflow at base (not top), with shutoff valve -Increased slope on sides, and riprap or sandbags added at shores would prevent water short circuiting and deter wildlife access
		Soils -Use substrate with less total and leachable metals and metalloids (especially copper) -Higher sand content would improve hydrology, constructability (ability to level soils, ease of planting) and accessibility for sampling -Organics should be mixed in bulk to soils prior to adding to cells
Assess commissioning timelines	Allow for proper phasing of implementing full-scale systems for closure	**Water** -Copper treatment improving through commissioning period; wetland achieving better treatment than suggested by inflow and outflow concentrations of system, as soils are leaching copper -Cadmium and selenium are also being removed from water -Wetland is maturing as expected and is performing beyond anticipated from the design -Tracer study recommended for 2016 to assess hydrology and pore volume of CWTS and determine HRT and removal rate coefficients for full-scale sizing

Objective	Purpose	Key Findings
		Soils -Soil redox has decreased as expected, reaching targeted ranges in Series 1 by the end of 2015, while Series 2 continues to establish -Significant amounts of metals are leaching from soil substrate into water, putting additional treatment demands on system
		Microbes -Sulphide-producing bacteria needed for copper and other metals removal have increased over time as soil redox achieved target ranges. Proportions are comparable to those in pilot system at similar point in commissioning -Abundance of selenium- and nitrate-reducing organisms are similar to those in pilot testing, indicating maturation as expected -Selenium treatment performance expected to increase as mosses continue to grow, as they can sorb dissolved selenium and harbour highest abundance of selenate-reducing microorganisms to render the selenium insoluble
Carex aquatilis transplantation effectiveness	Determine if plant propagation and/or replanting schedule will be needed for full-scale systems	->95% survival from transplanting -Within first 2 months a further increase of >20% -Full-scale system could be planted more densely to bring online faster, or less densely if time is less of an issue than sourcing plants (the plants are vigorous and will fill in the wetland in due time)
Moss colonization/distribution		-100% survival from transplanting -Slower to spread, needs to be started more densely -Staking helps maintain moss in 'upstream' parts of wetland, or could be transplanted multiple times through commissioning period

8. Tentative Schedule and Action Items for 2016

Based on progress of commissioning and early performance results in 2015, an action plan for CWTS optimization and testing for 2016 has been developed (Table 7). An updated multi-year schedule as per the proposed scope of work in the Minto Demonstration Scale Report Document 011_0315_01A (Contango, March 2015) is provided in Appendix C.

Table 7. Minto 2016 CWTS Demonstration-scale Action Items

Who/When	Task	Contango Action Required	Additional Information
Minto Staff, prior to first site visit	Add sandbags to perimeter of CWTS to minimize short circuiting	Provide diagram of sandbag location	Use sand that was confirmed to be good borrow source in 2016 (i.e., low copper)
	W15 creek monitoring	Develop research plan and schedule of testing	This is the creek(s) feeding the W15 area pond
Site visit 1 (May 2016)	Tracer study (salt)	Experimental design	Ensure salt tracer study will not affect other planned activities at the site (e.g., hydrogeology studies)
	Change AgCl probes	Replace old probes with new ones	These are the reference probes for soil-redox testing
	Develop flow rates for 2016	Based on tracer study	Flow rates will be set (and varied) according to expected flows at water storage pond in closure, and theoretical hydraulic retention times needed to develop outflow water quality predictions
	Check on W15 creek monitoring	Confirm monitoring locations	Because at least 3 entry points exist from the creek to the W15 pond, check which creek is being monitored and how this compares to the W15 sampling point water quality
Site visit 2 (July 2016)	Add organic material to System 2	Determine amount and type to add	This will only be performed if soil redox conditions have not consistently met targeted ranges by July 2016
Site visit 3 (September 2016)	General follow up	Final sampling for year	

9. Monitoring plan

A conceptual long-term testing plan was developed for the demonstration-scale CWTS and has been refined and adapted based on performance and scientific findings (Appendix A). The demonstration-scale wetlands are expected run until at least the end of 2016, and ideally longer in order to assess performance under a wider range of conditions. The conditions that could eventually be tested include both natural/environmental and selected influenced pressures, and can be imposed on the systems to mimic peak flow rates or droughts. In 2016 the systems are expected to continue to mature (i.e., complete commissioning), with plants becoming more established and abundant, and microbial communities accordingly acclimating to the targeted conditions. We expect to achieve targeted ranges of soil redox in 2016, which will indicate that the commissioning period has completed and allow for the monitoring program to shift focus to testing of operations and performance. A multi-year plan is provided in Appendix C, and includes work performed to date as well as a schedule for 2016 and potential activities for 2017.

10. Closure

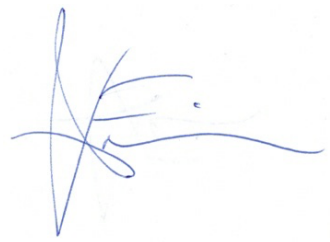
We trust the information herein satisfies your present requirements. Should you have any questions, please contact the persons listed below. We appreciate the opportunity to provide the services detailed in this report, and look forward to discussing any comments you may have.

Respectfully submitted,

Contango Strategies Ltd



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Appendix A

Table 1. Summary of analytical sampling types, frequencies and locations

Water	
Dissolved and total metals	Weekly, outflow of each cell and feed
Flow rate	Weekly, feed
pH, DO, ORP, Conductivity (in situ)	Weekly, all cells and feed
Anion Sum	Monthly, outflow of each cell and feed
Cation Sum	
Hardness ($CaCO_3$)	
Ion Balance	
Total Dissolved Solids (TDS)	
Chloride (Cl)	
Sulphate (SO_4)	
Nitrate	
Nitrite	
Nitrogen (Total)	
Alkalinity	
pH	
Conductivity	
Chemical Oxygen Demand (COD)	
Total Organic Carbon (TOC)	
Ammonia	
Total Kjeldahl Nitrogen (TKN)	
Total Suspended Solids	
Biological Oxygen Demand (BOD)	
Bicarbonate (HCO_3)	
Carbonate (CO_3)	
Hydroxide (OH)	
Soil	
Relative soil redox (in situ)	Monthly, all probes (6 per cell)
Cation exchange capacity (CEC)	Seasonally (3x per year)
SAR, pH, EC, %sat, Ca, F, Mg, Na, K, Cl, SO4	
Available NPK and sulphur	
Alkalinity	
Bicarbonate (HCO_3)	
Carbonate (CO_3)	
Hydroxide (OH)	
Total Organic Carbon (TOC)	
Dissolved Organic Carbon	
Metals Analysis (Total)	
Metals Leachable (SPLP)	

Ammonia	
Nitrate	
Nitrite	
Total Dissolved Solids	
Anion Sum	
Cation Sum	
Cation/EC ratio	
Ion Balance	
Sequential Leaching (5 Acid Test)	
Cation exchange capacity (CEC)	
Plant tissue samples	
Metals Analysis	*Carex aquatilis* and aquatic moss, each cell, year end
Microbial samples	
Growth-based most-probable number analysis (nitrate reduction, selenite and selenate reduction, total heterotrophs)	Seasonally (3x per year)
Genetic sequencing analysis for bacterial community composition and distribution	Seasonally (3x per year)

Appendix B – Copper Sequential Leach and Aluminum Graphs

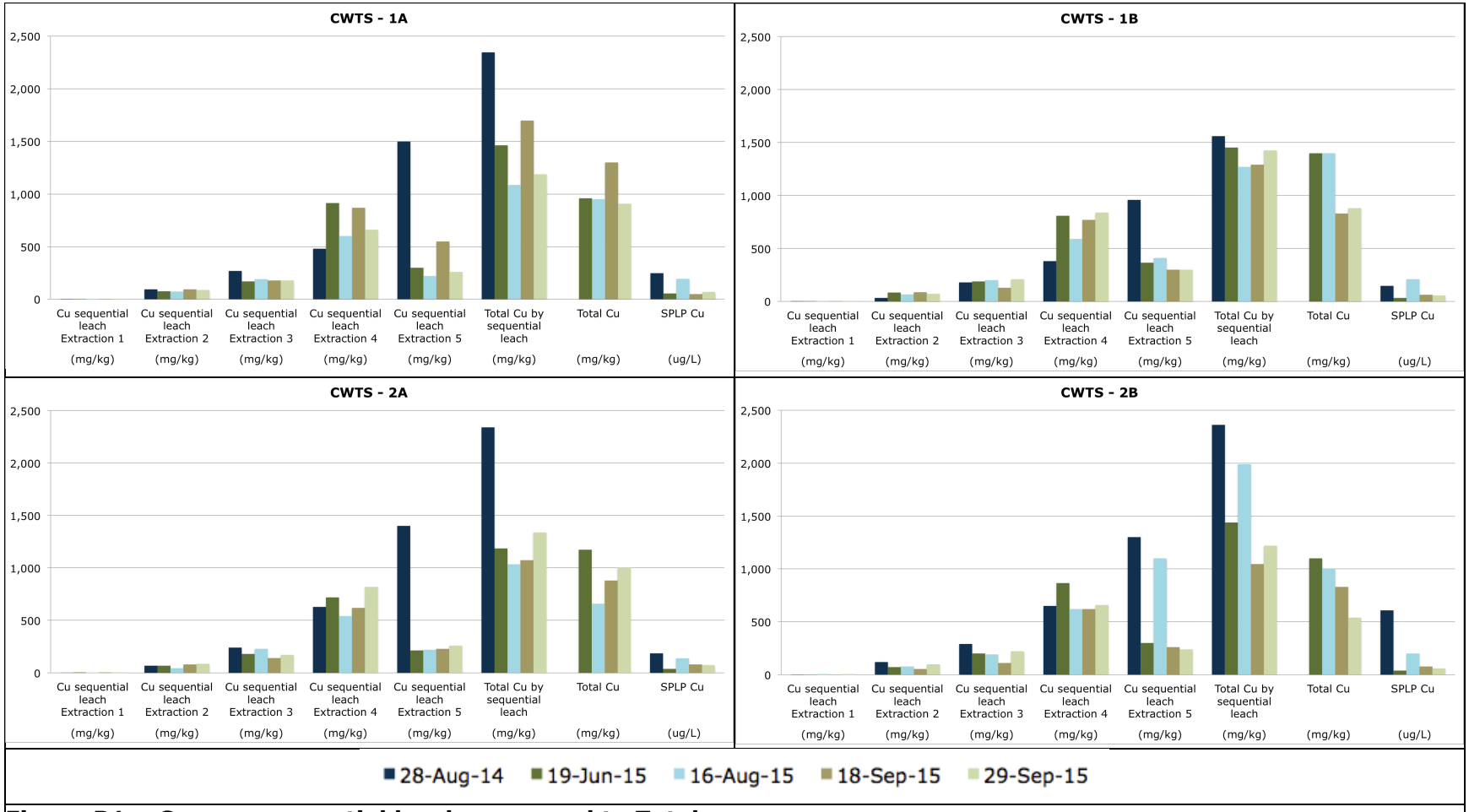


Figure B1 – Copper sequential leach compared to Total copper

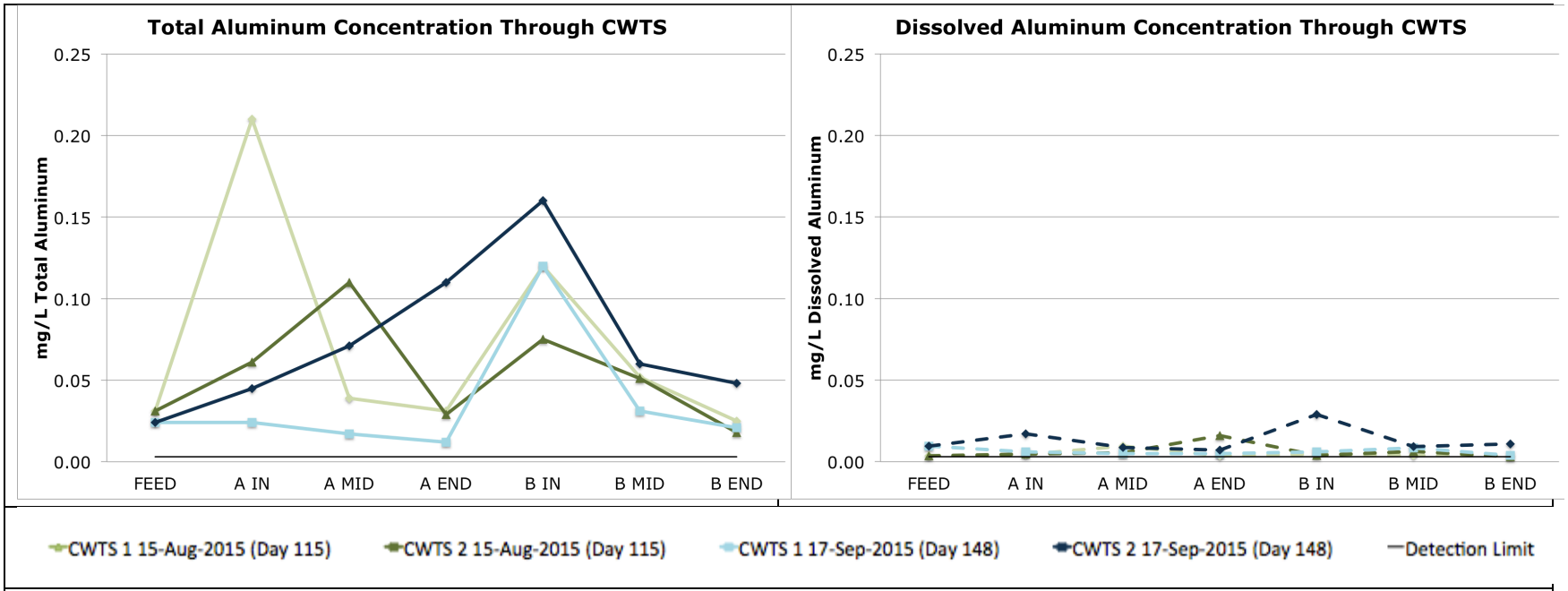


Figure B2 – Concentrations of Aluminum in water through the CWTS.
Water samples were collected from the CWTS feed, and from the beginning, middle, and end of each wetland cell on 2 occasions.

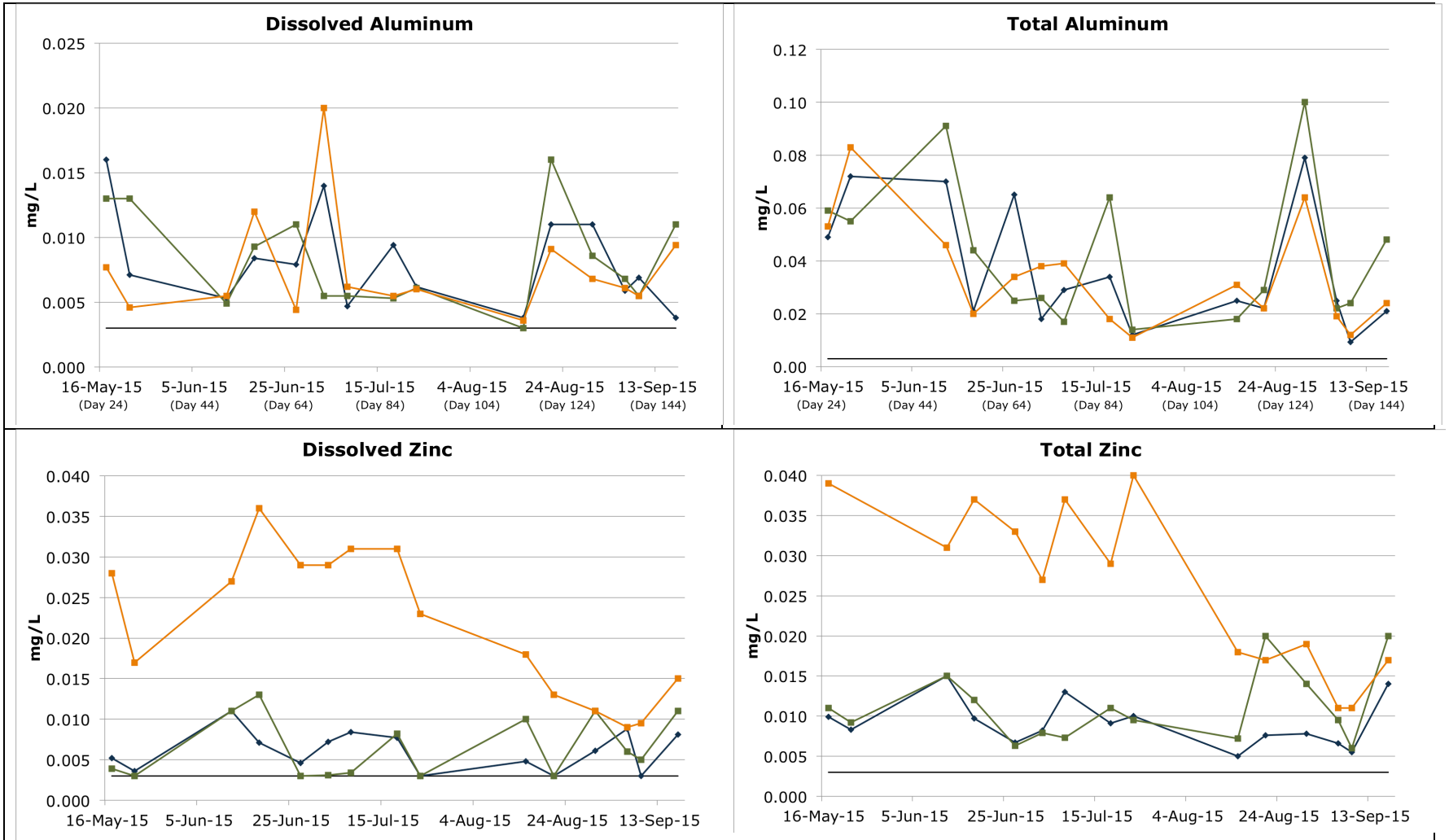


Figure B3 – Aluminum and Zinc concentrations in the demonstration scale wetland from May 17, 2015 to Sept 17, 2015.

The Measurement Uncertainty (MU) for aluminum and zinc is +/- 0.003 mg/L.

Appendix C

Table C1. Schedule of work

Item	Date	Activities	Actual
Construction	June 1-14 2014	Identify potential location for demonstration scale CWTS (CSL site visit – 1 scientist)	Completed
	June - July 2014	Engineering and geotechnical (Minto)	Completed
	July 2014	Construction (Minto)	Completed
	August 2014	Planting and bringing system online (CSL site visit – 1 scientist, 1 technologist), coordinate for local students to assist.	Completed (no students available, brought 2 technologists)
Commissioning	2014	Acclimation and maturation at constant flow rate, ~20 hr HRT	Completed
		September - CSL site visit/checkup (1 technologist, 1 scientist)	Did not occur because construction was last week of August.
	2015	Continued commissioning. Operation at constant flow rate, ~20 hr HRT	Completed (at shorter HRT)
		Spring – CSL site visit/checkup (1 technologist, 1 scientist), includes micro sampling	Completed
		Summer - Increase depth from 10 cm to 20 cm (1 technologist), includes micro sampling	Completed (scientist)
		Fall – CSL site visit/checkup (1 technologist), includes micro sampling	Completed (scientist)
Performance Monitoring	2016	Minto to add sandbags prior to first site visit and begin W15 creek monitoring	Expected May/June 2016
		Spring – CSL site visit (1 scientist, 1 technologist), includes microbial sampling and tasks outlined in Table 7 of report	Expected May/June 2016
		Summer - CSL site visit/checkup (1 scientist, 1 technologist), includes microbial sampling and tasks outlined in Table 7 of report	Expected July/August 2016
		Fall - CSL site visit/checkup (1 technologist), includes microbial sampling and tasks outlined in Table 7 of report	Expected September 2016
	2017*	Continued performance monitoring	*Putative activities, may not be necessary depending on performance and outflow water quality in other years.*
		Possible testing of semi-passive hybrid Bioreactor/CWTS phase. This will involve adding solid organic matter (such as alfalfa hay, straw) to the CWTS cell(s).	
Reporting	2014-2016	Reporting will be performed annually, with verbal and/or emailed interim updates	On Schedule

Appendix C

Table C1. Schedule of work

Item	Date	Activities	Actual
Construction	June 1-14 2014	Identify potential location for demonstration scale CWTS (Contango site visit – 1 scientist)	Completed
	June - July 2014	Engineering and geotechnical (Minto)	Completed
	July 2014	Construction (Minto)	Completed
	August 2014	Planting and bringing system online (Contango site visit – 1 scientist, 1 technologist), coordinate for local students to assist	Completed (no students available, brought 2 technologists)
Commissioning A	2014	Acclimation and maturation at constant flow rate, ~20 hr HRT	Completed
		September - Contango site visit/checkup (1 technologist, 1 scientist)	Did not occur because construction was last week of August
	2015	Continued commissioning. Operation at constant flow rate, ~20 hr HRT	Completed (at shorter HRT)
		Spring – Contango site visit/checkup (1 technologist, 1 scientist), includes micro sampling	Completed
		Summer - Increase depth from 10 cm to 20 cm (1 technologist), includes micro sampling	Completed (scientist)
		Fall – Contango site visit/checkup (1 technologist), includes micro sampling	Completed (scientist)
	2016	Minto to add sandbags prior to first site visit and begin W15 creek monitoring	Completed May/June 2016
		Spring – Contango site visit (1 scientist, 1 technologist), includes microbial sampling and tasks outlined in Table 5 of report. HRT tracer study completed, outlined in section 7 of report.	Completed June 2016
		Summer - Contango site visit/checkup (1 scientist, 1 technologist), includes microbial sampling and tasks outlined in Table 5 of report. Evapotranspiration study completed as outlined in section 9 of report. Organics were added to the CWTS as outlined in section 5.1 of report.	Completed July 2016
Commissioning B[1]		Fall - Contango site visit/checkup (1 technologist), includes microbial sampling and tasks outlined in Table 5 of report	Completed September 2016
	2017	Continue commissioning B. Operation at constant flow rate developed based on 2016 HRT tracer study.	Expected May 2017
		Spring – Contango site visit (1 scientist, 1 technologist), includes microbial	Expected May/June 2017

		sampling and tasks outlined in Appendix A of report. Proposed HRT tracer study. Sample for AVS to assess if the added organics are having the desired effect.	
		Summer - Contango site visit/checkup (1 scientist), includes microbial sampling and tasks outlined in Appendix A of report. Proposed evapotranspiration study.	Expected July/August 2017
		Fall – Contango site visit/checkup (1 scientist), includes microbial sampling and tasks outlined in Table 5 of report.	Expected September 2017
		Monitor soil redox. Determine when it consistently reaches targeted range.	Throughout 2017
Performance monitoring	*2018*	Fluctuate flow rates weekly, based on expected amount of water for full-scale wetland (scaled to size). Evaluate performance of wetland at different key periods.	*Putative activities, may not be necessary depending on performance and outflow water quality in other years.*
	2014-2018	Reporting will be performed annually, with verbal and/or emailed interim updates	On Schedule

[1] The commissioning-B period will continue in 2017 until the demonstration-scale CWTS performance is consistently achieving performance expectations.

Contango STRATEGIES LTD.

Forward looking. Lateral thinking.

Minto Mine Constructed Wetland Treatment Research Program – Demonstration Scale 2016 Update

Document – 011_0217_05B

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Executive summary

The Minto Mine, operated by Capstone Mining Corp., is located 240 km northwest of Whitehorse on the west side of the Yukon River. In 2013 Contango Strategies Ltd. (Contango) was retained by Capstone Mining Corp. to conduct a phased assessment to determine the feasibility of using passive means to treat water at closure at the Minto Mine. Pilot-scale testing, completed by Contango in July 2014, determined that plants and beneficial microbes found at the Minto site could be used to remediate contaminated mine drainage in a constructed wetland treatment system (CWTS; November, 2014). In August 2014, a demonstration-scale CWTS was implemented at the Minto Mine site. The performance results from the on-site demonstration-scale CWTS confirm that a passive water treatment system can be used to successfully improve mine-impacted waters.

To date, the demonstration-scale CWTS has had 312 operational days (i.e., water flowing), consisting of 24 days in 2014, 137 days in 2015, and 152 days in 2016. Day 248 (July 28, 2016) is when organics were added to the CWTS to aid in treating the additional copper load originating from the soils used as substrate, and marked the end of what we will now be referring to as the period "commissioning-A", and the beginning of period "commissioning-B". The commissioning-B period will continue in 2017 until the demonstration-scale CWTS performance is consistently achieving performance expectations. These commissioning periods are necessary for operational adjustments to be made (e.g., raising water depth, modifying outflow patterns), and for plant and microbial populations to establish and mature to achieve full operational performance of the CWTS. Owing to the types of soils used as substrate in construction (i.e., high concentrations of oxidized copper), commissioning has taken longer than would normally be the case (or would be the case for the full-scale system).

The commissioning period is also being used to collect additional information about the on-site demonstration-scale CWTS that will aid in the final full-scale CWTS design, sizing, and construction. For example, a tracer study was carried out in June 2016 to assess the hydrology and hydraulic retention time (HRT) of the CWTS. Using the data from the tracer study, the pore space involvement of the CWTS substrate was calculated.

Performance continues to be monitored and while full treatment functionality of the demonstration-scale system is not expected during the commissioning period, the information would aid in the development of the full-scale system in the following ways:

- Evaluation of construction effectiveness and potential optimizations.
- Assessment of timelines to reach targeted operational parameters to allow for effective phasing of implementation (e.g., soil redox, percentage and abundance of sulphate reducing bacteria).
- Assessment of the effectiveness of *Carex aquatilis* and aquatic moss transplantation to assess planting density, time period to full density, and if plant propagation and/or a replanting schedule is necessary.

In July 2016 the demonstration-scale system was not achieving stable soil redox values within the targeted range (nor associated performance expectations); therefore, woodchips and straw were added to the demonstration-scale CWTS as sources or organic carbon. Because

the demonstration-scale system takes time for plants to mature (and generate organic carbon for the microbes to feed on), the addition of organics early in the operation of the system can supplement the carbon sources needed for beneficial microbes that drive treatment of selenium and other constituents. This will be included in the full-scale system design specifications. The addition of organics marked the end of the commissioning-A period.

Once the commissioning–A period was complete, the commissioning-B period began on July 29 and continued to the end of 2016 operations on September 30, 2016. During the commissioning-B period, the demonstration-scale CWTS achieved on average 64% removal of dissolved cadmium (mean influent 0.0185 µg/L, mean outflow 0.0066 µg/L), 37% removal of dissolved copper (mean influent 46.1 µg/L, mean outflow 28.8 µg/L), a 21% removal of molybdenum (mean influent 7.3 µg/L, mean outflow 5.7 µg/L), a 41% removal of dissolved selenium (mean influent 5.7 µg/L, mean outflow 3.3 µg/L) and a 69% removal of dissolved zinc (mean influent 37.6 µg/L, mean outflow 11.6 µg/L). The percent removal of molybdenum and selenium is notable, as it has increased from 0% removal during commissioning-A, to achieving 21% and 41% removal, respectively during commissioning-B. Monitoring through 2015 and 2016 has indicated that percent removal of elements of concern have been increasing over time. The results of the 2016 on-site demonstration-scale CWTS performance indicate that the CWTS is continuing to mature. Plans for 2017 will focus on testing operations and performance. Flow rates will be set at a constant flow to evaluate the stability of the CWTS performance over one year. A second HRT tracer study will be performed using an updated method based on the findings from the 2016 HRT tracer study. A second evapotranspiration study will also be performed to determine the amount of water that is being lost to evapotranspiration from the CWTS. Weekly monitoring will continue throughout 2017 to evaluate performance of the CWTS.

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Abbreviations and definitions

Acidity – A measure of the capacity of water to neutralize base. Influences pH along with alkalinity.

Alkalinity (Alk) – A measure of the capacity of water to neutralize acids. Influences pH along with acidity.

Amendment – A chemical or organic material added to encourage specific conditions (e.g., aerobic/anaerobic, pH, ORP) or as a source of something that is needed for passive treatment (e.g., nutrients, alkalinity, binding sites, etc).

ANAMMOX – Anaerobic ammonium oxidation; a microbiological process where ammonium and nitrite are converted directly to dinitrogen gas ($NH_4^+ + NO_2^- \rightarrow N_2 + 2H_20$).

Biofilm(s) – A visible and often slimy-looking film (community) of bacteria (may contain other microbes).

Carbon source – A source of carbon (energy/electrons) for microbes (see **electron donors**). Examples include ethanol, methanol, acetate, sugar (glucose), molasses, wood chips, detritus (dead plant matter).

Carex aquatilis – A plant (emergent macrophyte) commonly known as water sedge.

Carex utriculata – A plant (emergent macrophyte) commonly known as Northwest Territory sedge or beaked sedge.

Contango – Contango Strategies Ltd.

Constructed wetland treatment system (CWTS) – Wetlands that are designed and constructed to remove compounds from water, using natural processes to sequester them into the soils rendering them less bioavailable. They are different from wetlands that provide habitat for wildlife.

Denitrification – Where nitrate (NO_3^-) is reduced by microorganisms to form nitrite (NO_2^-), nitric oxide (NO), or nitrous oxide (N_2O), or nitrogen gas (N_2) (see **nitrate reduction**).

Dissolved oxygen (DO) – Diatomic oxygen (O_2) dissolved in water; oxygen can dissolve in water by diffusion from surrounding air, as a product of photosynthesis, or through forced aeration.

Electron donor(s) – A chemical compound that donates electrons to another compound (see **carbon source**). An electron donor is a reducing agent, which that by virtue of it donating electrons, is itself oxidized (see **oxidation** and **reduction**).

Explanatory parameters – Quantifiable parameters that describe how water treatment reactions take place.

Evapotranspiration – The combined effects of open water evaporation and plant transpiration (Beebe *et al*, 2014).

Genetic analysis – Analysis to assess the presence, identity, and diversity of different microbes in a sample.

ICP-MS – Inductively coupled plasma mass spectrometry.

Macrophytes – An aquatic plant, large enough to be seen by eye. Can be emergent, submergent, or floating.

Microbes – Microscopic organisms that can be uni- or multi-cellular. This includes algae, bacteria, fungi, viruses, and yeast.

Most probable number (MPN) – A statistical value representing the viable population of microbes in a sample through use of dilution and multiple inoculations.

Oxidation – The loss of electrons, or increase in valence state, by a molecule, atom, or ion. Can be driven by microbes. Process is complementary to chemical reduction.

Oxidation-reduction potential (ORP) – A measure of the tendency of a chemical species to acquire or donate electrons, thus becoming reduced or oxidized, respectively, measured in millivolts.

Passive or semi-passive treatment system(s) (PTS) – General term used to refer to both passive and semi-passive treatment systems that use processes coupling transformations (e.g., chemical and biogeochemical reactions) with physical transfers (e.g., sorption, filtration) to remove constituents from water, often operationally passive with little long-term management required.

Redox – Oxidation-reduction potential (in sediment), a measure of the tendency of a chemical species acquire or donate electrons, thus becoming reduced or oxidized, measured in millivolts. This measurement is relative to the water ORP.

Reduction – The gain of electrons, or a decrease in valence state, by a molecule, atom, or ion. Can be catalyzed by microbes. Process is complementary to chemical oxidation.

Simultaneously extracted metals (SEM) – Amounts of heavy metals such as Ni, Zn, Cd, Pb, Cu, Hg in sediment, assessed in the context of AVS for excess sulphide. (also see **acid volatile sulphide**)

Sorption – The physical and/or chemical process by which one substance becomes attached to another substance.

Specific conductivity (SPC) – A measurement of electrical conductivity in water that is typically expressed in µs/cm, which has been adjusted for temperature (25°C).

Species (sp.) – One of the basic units of biological classification and a taxonomic rank. Rank in the classification of organisms below genus and above strain. Also can be used to refer to the oxidation state of a mineral (e.g., selenate and selenite are species of selenium).

SPLP – Synthetic precipitation leachate procedure.

Sulphide – An inorganic anion of sulphur that can form stable complexes with metals and make them insoluble in water (remove them from the water).

Sulphide producing bacteria (SPB) – Microbial reduction of sulphur compounds, such as sulphate, sulphite, thiosulphate, and sulphur, which produces sulphides and alkalinity. (see also **SRB**).

Sulphate reducing bacteria (SRB) – A form of sulphide producing bacteria that specifically uses sulphate for reduction (see **sulphide producing bacteria**).

Sulphide production – Microbial reduction of sulphur compounds, such as sulphate, sulphite, thiosulphate, and sulphur, which produces sulphides and alkalinity.

Total dissolved solids (TDS) – A measure of the combined organic and inorganic salts dissolved in water.

Total organic carbon (TOC) – A measurement of the total organic carbons present in water.

Total suspended solids (TSS) – A measurement of all particles in water that are larger than 2 µm (anything smaller than 2 µm considered a dissolved solid).

Transfer – Processes that treat water by transferring a constituent to another location without changing its form. For example: absorption, adsorption, dilution, dispersion, filtration, precipitation (aqueous to solid), and volatilization.

Transform – Processes that change the chemical form or state of a constituent. For example: biodegradation, biotransformation, hydrolysis, ionization, oxidation, photolysis, and reduction.

1. Introduction and background

The Minto Mine, operated by Capstone Mining Corp., is located 240 km northwest of Whitehorse on the west side of the Yukon River. The Minto property lies within the eastern part of the Dawson Range, with elevations from 700 to 1,000 m; the landscape has rounded mountains intersected by broad valleys and drainages that are part of the Yukon River watershed.

The Minto Mine has been in commercial operation since October 2007 and the deposits being mined are copper sulphide mineralized zones. Surface and groundwater water quality is a key consideration in the evaluation of potential effects of mining and mineral development projects and changes to water quality parameters have the potential to affect aquatic and human use of water resources. A Reclamation and Closure Plan (RCP) is required under both the Water License and the Quartz Mining License. The RCP is intended to address the long-term physical and chemical stability of the site and closure of the proposed features and disturbances associated with the mine. As a part of the RCP, a Constructed Wetland Treatment System (CWTS) is being designed, evaluated, and optimized for water treatment at closure through a phased program (Minto Phase V/VI Expansion Project, YOR Project Number 2013-0100).

For CWTSs to be effective, they must be designed, piloted, optimized, implemented, and maintained in a site-specific manner. A scaled approach for CWTS implementation allows for improvement, optimization, and flexibility for modifications along each step. Phases include:

1) site assessment and information gathering,
2) technology selection and conceptual design,
3) pilot-scale testing and optimization (controlled environment),
4) on-site demonstration-scale confirmation and optimization, and
5) full-scale implementation.

Phases 1-3 have been completed (reports 2013-0100-256 and 2013-0100-257 on YESAB registry, and Contango, March 2014; Contango, November 2014) and confirmed plant amenability to transplantation and the CWTS design for further on-site testing. During pilot-scale trials, the selected CWTS design achieved on average 92% removal of cadmium (mean influent 0.336 µg/L, outflow 0.027 µg/L), 92% removal of copper (mean influent 146 µg/L, outflow 11.3 µg/L), 41% removal of selenium (mean influent 10.2 µg/L, outflow 6 µg/L), and 92% removal of zinc (mean influent 40 µg/L, outflow 3.2 µg/L), using synthetic influent designed to mimic the worst-case water chemistry of a long-term closure scenario, but tested under the controlled conditions (e.g., controlled flows, known temperature, etc) of an off-site treatability testing center (Contango, November 2014). It should be noted that lower influent concentrations will have a lower percent removal even when achieving the same final outflow concentrations.

Phase 4 of the project is underway, with the on-site demonstration scale CWTS constructed at the Minto Mine during fall 2014 (Contango, March 2015). The on-site demonstration-scale CWTS operated for 137 days in 2015 and 152 days in 2016 (Table 1). The results of the 2015 and 2016 on-site demonstration-scale CWTS performance indicate that the CWTS is maturing as expected and maturation continued to be monitored in 2016. This document reports on

the on-site demonstration scale CWTS data from construction, and through two years of commissioning-A (2015/2016) and into the commissioning-B period (late 2016).

Table 1 – Days of operation of CWTS.

Scale	Year	Days Operated	Date	
			Start	End
Pilot (off-site)[1]	**2013 -2014**	205	Nov 4	May 28
Demonstration (on-site) Commissioning-A	**2014[2]**	23	Aug 27	Sep 19
	2015	137	May 16	Sep 29
	2016	101	May 2	Jul 28
Demonstration (on-site) Commissioning-B	**2016**	51	Jul 29	Sep 30

[1] Pilot-scale system operated from Nov 4, 2013 to July 16, 2014.
[2]The system was constructed in 2014, but no water testing occurred during this first month of commissioning-A.

2. Construction and operation

2.1. System layout and dimensions

The demonstration-scale CWTS includes 2 systems in parallel with 2 cells in each series and a final catchment basin that both systems flow into (Figure 1). Dimensions and construction details are available in the Minto Demonstration Scale Report Document 011_0315_01A (Contango, March 2015). The two parallel systems serve as a replicate for data analysis, and as testing has progressed, the two systems have also allowed for comparison of different management techniques. Dimensions of the systems are provided here in Table 2.

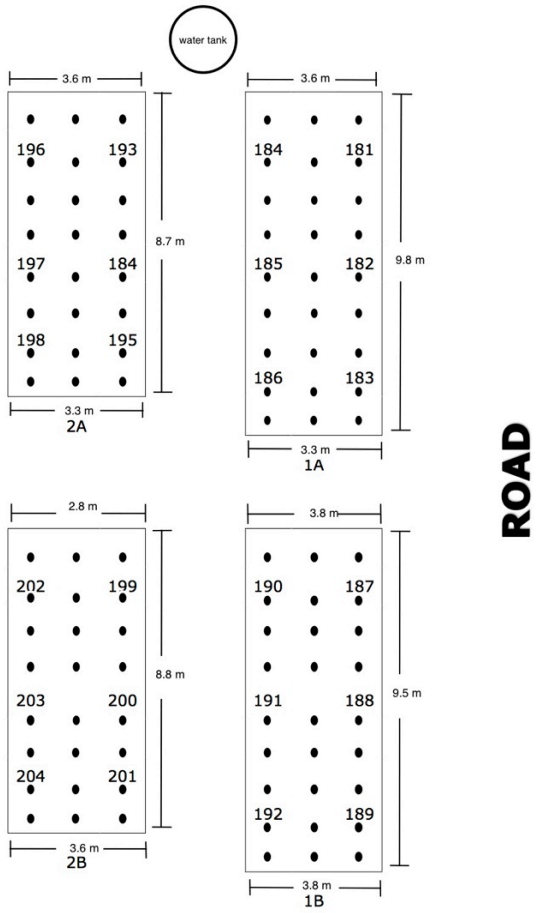


Figure 1 – Diagram of demonstration-scale CWTS.
Dimension measurements are indicated at soil surface. Black dots indicate initial construction grid marked by moss stakes, and locations of soil redox probes with identifying numbers.

Table 2 – Dimensions of demonstration-scale CWTS cells at soil surface and resultant areas of treatment systems.

Measurement		Series 1		Series 2	
		1A	1B	2A	2B
Width (m)	Inflow	3.6	3.8	3.6	2.8
	Outflow	3.3	3.8	3.3	3.6
Length (m)		9.8	9.5	8.7	8.8
Approximate surface area at soil (m^2)		33.8	36.1	30.0	28.2
Total area of system at soil (m^2)		69.9		58.2	

2.2.Substrate soils used in construction

Soils used to construct the CWTS are described in the initial report that outlines construction (Contango, March 2015). In brief, the recommended soil for the CWTS was sand, with 2-7% by volume as organic material (e.g., woodchips, peat). In the pilot-scale systems, this resulted in a total organic carbon (TOC) content by weight of 0.2-0.6% (the sand itself was at 0.1% TOC prior to adding amendment). Ideally, this percentage could be higher, approximately 2-10% by weight to stimulate the desired reducing conditions. For the demonstration-scale system, the soil added to each of the 4 cells during construction was from a local borrow site. As is expected in a mining area, the soils are likely mineralized. Although a potential borrow source was tested prior to construction ("Tested Soil"; Table 3), a different borrow source was available upon construction of the demonstration-scale wetland ("Soil used"; Table 3). The material used in the construction of the demonstration-scale wetland was an organic peat, and analyses received after construction indicated an elevated concentration of leachable copper (Table 3). It should be noted that in a full-scale system, the variability in soil substrate would be normalized by the larger volume of soil used. The substrate in the wetland (composed of the organic peat soil, wood chips, and straw) had a TOC content of 1.8-3.1%. The implications of this are discussed in section 6.1 of this report.

Table 3 – Comparison of copper concentrations in soils of pilot- and demonstration-scale CWTS.

Test method	Pilot-scale	Demonstration-scale	
	Initial soil	Tested soil (June, 2014)	Soil used (August, 2014)
SPLP Copper (mg/L)[1]	-	0.00546-0.0296[2]	0.148-0.608
Total Copper (mg/kg)	5.3-5.5	210-1400[2]	960-1400[3]

[1] SPLP - Synthetic Precipitation Leaching Procedure
[2] For the June 2014 samples, the soil with the highest total copper concentration also had the lowest leachable copper concentration, and was therefore deemed acceptable for use.
[3] Total copper value for soils used was taken in June 2015 (no data for August 2014).

2.3.Vegetation used in planting

The demonstration-scale CWTS was planted with *Carex aquatilis* (aquatic sedge) and aquatic mosses from the W10 area of the Minto Site. The plant selection and borrow source was previously determined through the site assessment (reports 2013-0100-256 and 2013-0100-257 on the YESAB registry) and pilot-scale testing (Contango, November 2014). Five *C. aquatilis* plants were planted per square meter, with moss tied to stakes that outlined the 1 m x 1 m grid for planting (details provided in Contango, March 2015).

2.4.Water source and flow

Water from the W36 area receiving seepage from the toe of the Mill Valley Fill Extension (MVFE) was selected for the demonstration-scale CWTS testing as the leachate is similar to that expected upon closure in the MVFE area. The chemistry of this water at the time of bringing the demonstration-scale CWTS online (September 18, 2014) is provided in Contango,

March 2015. In 2015 the water for the CWTS was pumped out of the sump at the toe of the MVFE. The Mill Valley Fill Extension Stage 2 (MVFE2) collection sump (W62 sump) was constructed from November 9, 2015 to February 7, 2016. The sump was constructed as a replacement for the sump used in 2015. The W36 sump was decommissioned in February, 2016 as part of the MVFE2 construction. The new sump approximately 30m downslope was the supply water for the CWTS for the summer of 2016 (Figure 2).



Figure 2 – Water source for the demonstration-scale CWTS in 2016.

After construction in 2014 and in early 2015, the flow rates for the systems were planned to be set to have a 10 day nominal hydraulic retention time (HRT), meaning water entering the wetland takes 10 days to exit. This is referred to as a nominal HRT because it is a calculation based on the size of the wetland and the amount of water entering, and not confirmed empirically using tracing dyes. This HRT is much longer than what is necessary to achieve treatment based on the pilot-scale systems (~3 days), and was chosen to facilitate plant establishment and maturation. However, due to the leachable copper concentrations in the soils used for construction, it was instead decided to run the systems at a faster flow to wash as much leachable copper from the soils as possible. As such, a shorter nominal HRT of ~20 hrs (Figure 3) was used as a starting point (HRT calculated using measured water depths of approximately 15 cm and negligible pore water involvement [30cm of soil at 10% pore volume] due to peat soils). Despite the shorter HRT resulting from the faster flow of water, the systems have acclimated and matured as was expected for the longer HRT. Because the sizes of the CWTS systems are slightly different, Series 1 (closer to the road) and Series 2

(further from road) are set with flow rates to result in similar HRTs in each system. Flow rates were monitored and adjusted throughout 2015 based on CWTS establishment and maturation (further information in sections 3 and 4).

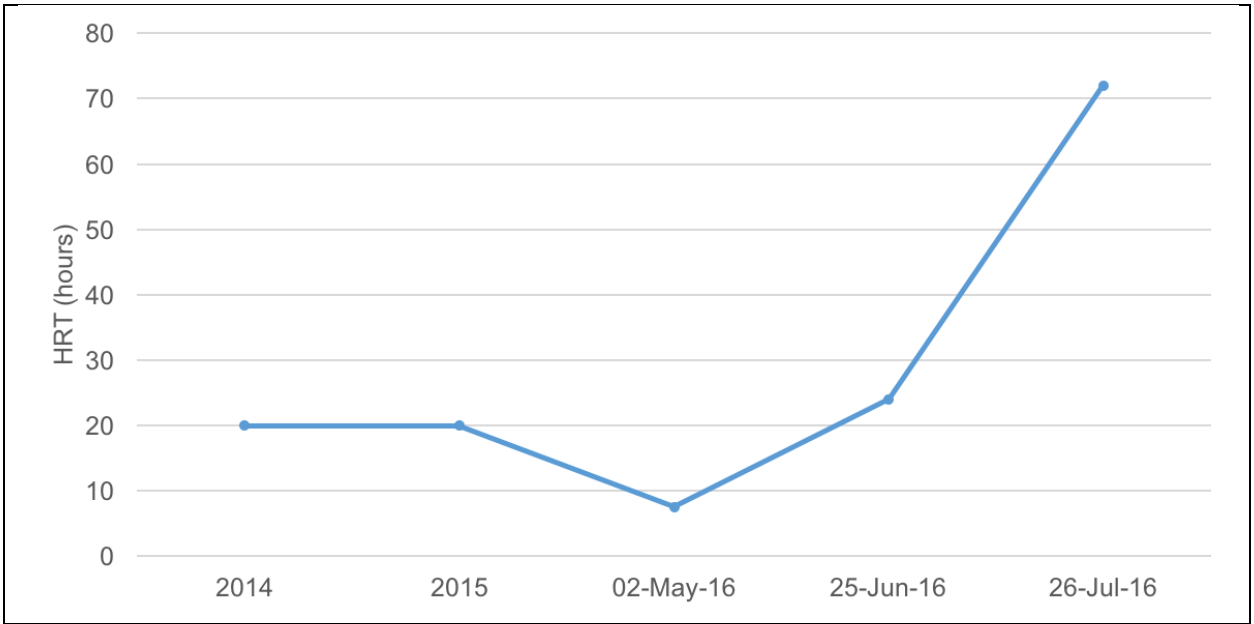


Figure 3 – Demonstration-scale CWTS targeted hydraulic retention time from construction to present.

Three flow rates were used during operation of the CWTS in 2016. At start up in May 2016 the nominal HRT was set to ~7.5 hours to further attempt to flush copper out of the soil (further enhancing the 2015 EDTA trials conducted in System 2 (section 6.3)). In early June, HRT was increased to ~24 hours to allow for a tracer study to be conducted during a 2-day site visit in attempt to determine the actual HRT (vs nominal, calculated HRT) of the CWTS (section 7). At the end of June, the HRT was lengthened to a calculated ~72 hours (3 days) for performance testing and remained at this HRT until the end of operations in September 2016. Due to pump and flow meter issues the actual flow rates varied from the target flow rates throughout 2016 operations. Table 5 outlines known pump and flow meter issues that occurred in 2016, and Figure 4 shows the fluctuation in actual vs targeted flow rates throughout the 2016 operations.

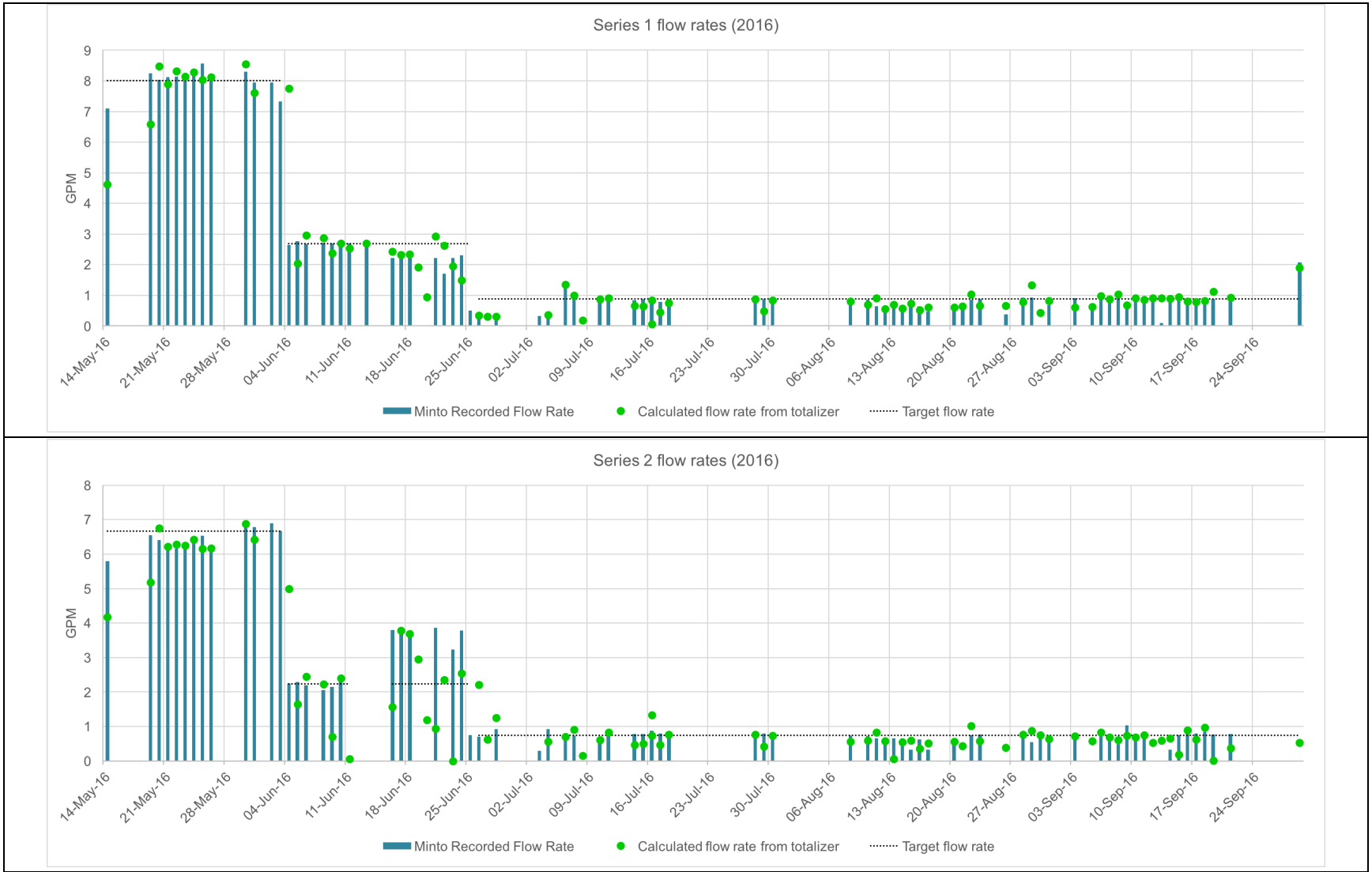


Figure 4 – Flow rates during CWTS operations in 2016.
Blue bars indicate flow rates displayed on meter, while green dots are flow rates calculated from the totalizer values (values from meter were recorded by Minto personnel. The totalizer value is the cumulative volume of water passing through the flow meter between two measurement dates. Areas with no blue bars or green dots are time periods where flow rates and totalizer values were not recorded.

3. Commissioning

The period between the construction of the CWTS and achieving the expected treatment performance is referred to as the commissioning period. During this period, the plant and microbial populations establish and mature, and construction effectiveness and optimization opportunities are evaluated to make operational adjustments (e.g., raising water depth, modifying outflow patterns) and inform on the full-scale system (Table 4). For the demonstration-scale CWTS the commissioning period was divided into two periods, commissioning-A, and commissioning-B. The end of the commissioning-A period and the beginning of the commissioning-B period was marked by the addition of organics on July 28, 2016.

Performance continues to be monitored and while full treatment functionality of the demonstration-scale system is not expected during the commissioning period, the information will aid in the development of the full-scale system in the following ways:

- Evaluation of construction effectiveness and potential optimizations.
- Assessment of timelines to reach targeted operational parameters to allow for effective phasing of implementation (e.g., soil redox, percentage and abundance of sulphate reducing bacteria).
- Assessment of the effectiveness of *Carex aquatilis* and aquatic moss transplantation to assess planting density, time period to full density, and if plant propagation and/or a replanting schedule is necessary.

However, there are aspects of the demonstration-scale CWTS that are not directly transferrable to the planned construction of the full-scale CWTS. Notably, the demonstration-scale CWTS was built with a different soil type and chemistry than recommended for the full-scale system. Therefore; it is taking a longer time to commission the demonstration-scale system as the copper in the soils needs to be treated in addition to the copper and other constituents in the water.

Based on pilot-scale testing (Contango, November 2014), the estimated commissioning period for the Minto demonstration-scale CWTS was 4 months of operation (i.e., period when water is flowing). In 2014, the demonstration-scale CWTS operated for less than one month prior to freezing, and was restarted along with spring thaw (freshet) in mid-May 2015. Based on these timelines, September 2015 was month 4 of commissioning, and the demonstration-scale system was trending towards performance as expected during the commissioning period (Contango, March 2016). However, as discussed further in sections 2.2 and 6.1, the soil used in the CWTS had high concentrations of leachable copper, and therefore optimization opportunities were evaluated throughout 2015, and adjustments were implemented in 2015 and 2016. The last operational adjustment during the commissioning period was the addition of organic material in 2016, which marked the end of the commissioning-A period and the beginning of the commissioning-B period. Organic material will also be added during construction of the full-scale design to enhance desired reducing conditions to establish in the CWTS until vegetation is sufficient to provide for this required component.

Table 4 – Operational adjustments made to the demonstration-scale wetland.

Year	Activity	Purpose
2014	Construction	Construction
2015	Increasing water depth	Plant acclimation and establishment
	Adding more aquatic moss	Increase moss density
	Modifying inflow and outflow distribution areas	Promote uniform flow field
	Removing *Carex utriculata*	Was erroneously planted
	Wrapping inflow tank	Prevent algal growth
	Fe-EDTA test on CWTS series 2	To test if non-invasive treatments could be used to further leach the metals and improve soil quality
2016	Adding sandbags to the perimeter of the CWTS	Promote uniform flow field and prevent short circuiting of water around perimeter
	Perform tracer study	Determine the actual hydraulic retention time (HRT) of the CWTS
	Increase water depth of Series 1	Bring to desired specification depth
	Evapotranspiration study	Determine the amount of water loss occurring (and therefore concentration of elements)
	Setting flow rates at a 3 day calculated HRT (compared to 20 hrs previously used)	Continue commissioning period and develop rate kinetics
	Adding organics (woodchips and straw)[1]	Enhance desired reducing conditions (will be added to full-scale design specifications)

[1] This marked the end of the commissioning-A period and the beginning of the commissioning-B period.

4. Timeline and sampling schedule

Major events and operational adjustments are listed in Table 5. The actual dates as well as relative days of operation are provided, which adjusts for time that the CWTS was not receiving water when it was frozen. The relative days of operation allows for comparison to expected timelines from the pilot-scale testing, and for planning and scheduling to be done for full-scale construction and commissioning in the context of mine closure planning.

The sampling schedules for 2015 and 2016 were conceptually developed prior to beginning construction of the demonstration-scale CWTS (Appendix A). Actual dates of sampling were dependent on timing of spring thaw and winter freeze-up, and the associated ability to have the pumps operating at the W62 sump to supply water to the demonstration-scale CWTS.

Table 5 – Events and sampling activities since construction.

Event	Key Activity	Flow Rate Setting m^3/day (gal/min)		Calendar Date	Day(s) of Operation
		CWTS Series 1	CWTS Series 2		
CWTS constructed and planted	First sampling, water started.	-	-	August 27 – 31, 2014	0-4
Freeze up for winter	Feed water pumps turned off.	-	-	September 19, 2014	23
Winter 2014/2015					
Start up for 2015	Feed water pumps turned on.	14.17 (2.60)	11.61 (2.13)	May 16, 2015	24
Contango Site Visit #1	Microbiology, soils, water tested. *Carex* stem counts. Added more aquatic moss. Put black wrap on water tank to prevent algal growth. Water depth adjusted with sandbags.			June 18, 2015	57
Flow rate increased	Flow rates increased.	17.44 (3.20)	15.81 (2.90)	July 13, 2015	82
Contango Site Visit #2	Microbiology, soils, water tested. Added more aquatic moss.			August 16, 2015	116
Contango Site Visit #3	Water tested. Started Fe-EDTA test on System 2.			September 17, 2015	148
	Microbiology, soils, plants, water tested.			September 18, 2015	149
Fe-EDTA Testing	Daily total and dissolved copper analysis conducted at Minto.			September 19 – 26, 2015	150-157
Freeze up for Winter	Feed water pumps turned off.	-	-	September 29, 2015	160
Winter 2015/2016					
Flow meters installed and pumps turned on	Flush copper leaching from soil (associated with 2015 EDTA test).	44.10 (8.09)	38.10 (6.99)	May 2, 2016	161
Start up for 2016	Feed water pumps turned on.			May 13, 2016	172
Pump problems	Getting proper pumps installed at new sump location.			May 13-17, 2016	172-176
Flow rate decreased	HRT increased to 24 hours.	14.39 (2.68)	12.26 (2.24)	June 4, 2016	194
Added sandbags	Added sandbags to perimeter of CWTS.			June 7, 2016	197
Contango Site Visit #4	Microbiology, soils, water tested.			June 13 – 15, 2016	203-205
	Tracer test for HRT.			June 13 – 17, 2016	203 - 207

Table 5 – Continued.

Event	Key Activity	Flow Rate Setting m^3/day (gal/min)		Calendar Date	Day(s) of Operation
		CWTS Series 1	CWTS Series 1		
Inconsistent flow	Flow rates in series 2 inconsistent and above targeted flow rate.	14.39 (2.68)	20.71 (3.80)	June 16 - 24, 2016	206-214
Flow rate decreased	HRT increased to 72 hours.	2.73 (0.89)	4.09 (0.75)	June 25, 2016	215
Flow meter switched on series 1	Max flow on new flow meter is lower than targeted flow rate due to flow meter being installed backwards.	Unknown		June 25 – July 4, 2016	215-224
Flow meter for series 1 was on backwards resulting in unknown flow rates	Installed the flow meter correctly and reset flow rate to meet targeted HRT range of 72 hrs.	(0.89)		July 4, 2016	224
Evaporation Study	Flow stopped.	-	-	July 18 – 26, 2016	238-246
End of evaporation study	Flow turned back on.	2.73 (0.89)	4.09 (0.75)	July 26, 2016	246
Contango Site Visit #5	Microbiology and water tested.			July 27 - 28, 2016	247-248
	Added organic material (straw and woodchips)			July 28, 2016	248
Flow interruptions	Flow meters plugged.			August 15 –18 2016	266-268
Contango Site Visit #6	Microbiology, soils, water, and plants tested.			September 7 – 8, 2016	289-290
Freeze up for Winter	Feed water pumps turned off.	-	-	September 30, 2016	312

5. Monitoring explanatory parameters

Explanatory parameters are quantifiable aspects of a CWTS environment that can be used to assess feasibility of treatment for a range of constituents, and therefore 'explain' the performance of a CWTS. These parameters, which often include acidity, alkalinity, conductivity, dissolved oxygen (DO), pH, oxidation reduction potential (ORP), ion balance, available electrons donors (e.g., organic carbon, reduced elements), and temperature, can be used to predict, promote, and/or optimize the ability of the system to treat different constituents (Haakensen et al., 2015). A YSI ProPlus meter was used in the field to test for water temperature, DO, conductivity (and specific conductivity; SPC), pH, and ORP.

Average water temperature of the demonstration-scale CWTS in 2015 was 12.9°C, ranging from 0.5°C to 25°C, and in 2016 was 10.2°C, ranging from 0.8°C to 23.2°C. As would be expected, both the month of testing and time of day were found to affect temperature variation (section 7). DO concentrations in the CWTS cells were on average 10 mg/L in 2015, 15.9 mg/L in 2016 prior to commissioning being complete, and 8.4 mg/L in 2016 after commissioning was complete, which is higher than the pilot-scale systems (average 4.8 mg/L). This higher DO concentration makes the system more oxidizing and therefore more difficult to carry out nitrate and selenium treatment, and produce sulphides for copper and cadmium treatment. However, the conductivity, pH and ORP of the demonstration-scale CWTS cells are all very similar to the pilot-scale systems (Table 6).

Table 6 – Average in situ measurements from the pilot scale and demonstration-scale system.

Testing Period		DO (mg/L)	SPC (µS/cm)	pH	ORP (mV)	Soil redox (mV)
Pilot 2014[1]		5.73	990.0	7.71	159	-185
Demonstration-scale 2015		10	817.9	8.11	147.9	-52
Demonstration-scale 2016	commissioning-A[2]	15.9	890.9	7.79	143.7	-85
	commissioning-B[3]	8.4	1020	7.59	157.6	-89

[1] Pilot system data is from the low N system with moss and *Carex*. This was the system that was used as the foundational design.
[2] Data for commissioning-A period is from May 2-July 28, 2016.
[3] Data for commissioning-B period is from July 29-September 30, 2016.

An additional explanatory parameter that can be used to monitor maturation of the wetland during the commissioning period is the soil redox potential, which is measured using platinum tip probes (in soil) and Calomel electrodes (in water). This measurement offers insight into the direction of electron flux between the sediment/soil/pore water and overlying water column (Faulkner et al., 1989; Huddleston & Rodgers, 2008), and can be used to confirm reducing conditions in the soil. Based on the information gathered in pilot-scale testing, the targeted soil redox for the demonstration-scale CWTS is between -100 and -250 mV. In these redox ranges, bacterial sulphide-production through reduction of sulphur compounds (e.g., sulphate) is expected. Sulphide production directly results in metals and metalloid treatment

for elements such as cadmium, copper, molybdenum, nickel, lead, and zinc by precipitation as metal sulphides (Figure 5).

This maturation period is necessary for sufficient quantities of microbes to populate the wetland and become active in decomposing organic material. It is the electrons produced by the decomposition of organic material that is reported by the soil redox measurements. The decomposition of organic material then feeds the sulphate-reducing bacteria the type of energy they need to produce the sulphides that remove the copper, cadmium, molybdenum, and zinc from the water. The microbial activity of the system is discussed further in Section 10.3.

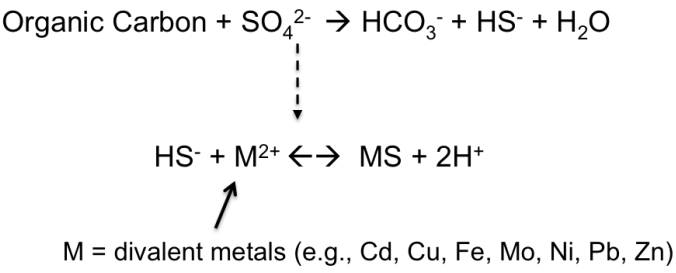


Figure 5 – Example equation of sulphate reduction and resultant metals mineralization and treatment.
Equation is not balanced; demonstrates sulphate reduction with creation of alkalinity and sulphides, which can then form metal sulphide complexes.

As expected from the pilot-scale testing, the soil redox in all the demonstration-scale CWTS cells has decreased over time, indicating maturation of the system (Figure 6). At the end of 2015, only Series 1 had begun to achieve targeted soil redox values that are conducive to sulphide production (ahead of anticipated schedule). It is possible that Series 1 had more organic material in the soils than Series 2, because of how construction occurred and this wasn't reflected in the small sample size sent for analytical testing.

Some soil redox probes were reporting negative values in the targeted range by July 18, 2015 (87 days); however, in general the CWTS took approximately 4 months of operation to establish generally reducing conditions. By September 17, 2015 (day 148), most of the soil redox probes within Series 1 were reporting redox values within the targeted range, with Series 2 still trending downwards but not yet within the targeted range. In comparison, the pilot-scale systems were stable and reducing within approximately 4 months of construction (Figure 6). In early 2016, the soil redox in both series was not being maintained within the targeted range, which indicated that additional organics may be needed in the CWTS to enhance maturation.

5.1. Addition of organics

Since the demonstration-scale system was not achieving stable redox values (nor associated performance expectations) within the targeted range, woodchips and straw were added to the

demonstration-scale CWTS on July 28, 2016. Because the demonstration-scale system takes time for plants to mature (and generate organic carbon for the microbes to feed on), the addition of organics early in the operation of the system can supplement the carbon sources needed for sulphate-reducing bacteria, and produce the desired sulphur- and selenium-reducing conditions (section 10.3). This will be included in the full-scale system design specifications.

One compressed 20-liter pail of woodchips was spread across the surface of each of the four CWTS cells (Figure 7). The A cells also each received one 0.16 m^3 bale of straw while the B cells received half of a straw bale of the same size. In the A cells, the straw was added along the inflow, outflow and along the sides of the cell in between the sand bags. In the B cells, straw was added at the inflow before and after the sand bags, and along the sides in between the sand bags (Figure 7).

By September 8, 2016 (final redox readings of 2016) redox values for cell 2B were consistently within the targeted range, and the redox values for series 1 were achieving targeted ranges 50% of the time (3 of the 6 in situ probes were reading within the targeted range). It is anticipated that the addition of organics will continue to provide the necessary carbon source to stimulate desired reducing conditions in 2017.

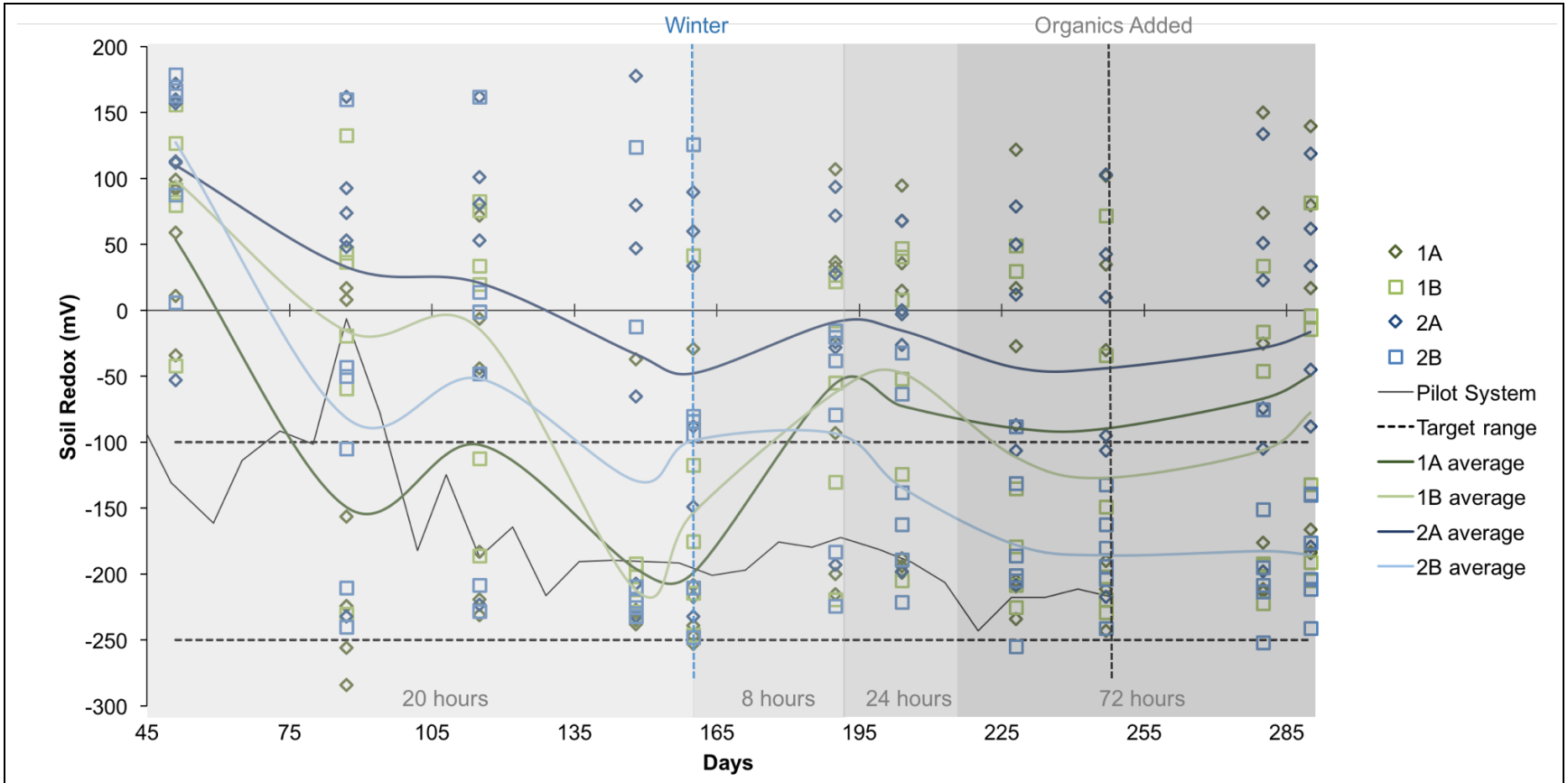


Figure 6 – Soil redox potential of each CWTS cell over time.
All demonstration-scale CWTS soil redox measurements are plotted, along with the average soil redox of the pilot-scale systems. Targeted soil redox values based on pilot-scale testing are indicated with dotted lines. The blue dotted line indicates break in measurements for winter 2015. Hours and associated grey shading indicate the HRT.



Figure 7 – Addition of organics on July 28, 2016.
Top picture shows cell 1A with woodchips that were added (floating on water surface). Bottom picture shows cell 2A with straw that was added after woodchips.

6. Performance progress during commissioning period and early operation

6.1. Soils

6.1.1. Methods

Although unintentional, the use of soils in the CWTS with high initial leachable copper concentrations (Table 3) allowed for additional types of testing to be carried out on these systems. Because the soil substrates used for construction of the CWTS were from overburden sources, the copper was not in a mineral form that would typically be found in a reducing CWTS (i.e., negative relative soil redox). Therefore, there has been some initial leaching of copper (and other elements, such as aluminum) from the soils into the water (Table 8, Figure 10, and Figure 12).

To assess the effect of the elevated metals (aluminum and copper) in the soils used in construction on CWTS functionality, four soil analytical test methods were used:

- Total concentration of elements in the soils by ICP,
- In 2015, leachable concentrations of elements in the soils by Synthetic Precipitation Leaching Procedure (SPLP), which is a method to assess the mobility of elements in soils at the pH of rain water (i.e., if the wetland were to entirely dry out, then be subjected to leaching by rain water) (EPA Method 1312/6010B)
- In 2016 a leach method was used. This analysis was carried out using a leaching procedure which involved the gentle tumbling of a sample in a specified leaching solution (water from the CWTS) for 2 hours. The resulting extract is then analyzed by inductively coupled plasma – optical emission spectrophotometry (EPA method 6010A). This method was selected in 2016 to determine the concentration of metals that are being released from the soil into the overlying water column.
- Sequential extraction procedure for the speciation of particulate trace metals (Tessier *et al*, 1979, EPA 6020A) to assess the stability and form of elements in soils (Table 7).

Table 7 – Summary of extractable fractions from sequential extraction procedure for the speciation of particulate trace metals[1].

Fraction	Description	Elements unstable when
1	Exchangeable fraction for adsorbed minerals	Readily released (i.e., soluble and exchangeable)
2	Mineral fraction bound to carbonates	Decreased pH
3	Mineral fraction bound to Fe-Mn oxides	Reducing conditions
4	Mineral fraction bound to organic matter and sulphides	Oxidizing conditions
5	Residual mineral fraction (primary and secondary minerals)	Not expected to be released in solution over time under conditions normally encountered in nature

[1] Method based on Tessier *et al.,* 1979.

6.1.2. Soil Aging

While the total copper concentrations have increased over time (with the addition of mineralized copper from the water), the leachable copper (measured by SPLP) in 2015 decreased over time. By day 58 of operation, the top 5 cm of soil was decreased to only 10-20% of the initial leachable copper concentration, and the lower 10-20 cm layers reached similar concentrations by day 149 (Table 8). In 2016 a new leach method was used to determine the concentration of metals that are being released from the soil into the overlying water column, rather than using the SPLP method which assumes leaching at rain water pH. The main difference between these methods is that the SPLP method uses laboratory water at the pH of rain water, while the new leach method uses a sample of the overlaying water in the CWTS. Therefore, the leach method is more representative of actual potential leachability of the soils into the wetland than is the SPLP method. In 2016 there was an overall decrease in leachable aluminum and copper from a depth of 0-10 cm (Table 8 and Appendix D). The B cells experienced a significant decrease of leachable copper of approximately 80% between June 15 and September 7, 2016. Cell 1A decreased 53%, and cell 2A had very little change in leachable copper. Cells 1A, 2A, and 2B experienced a similar decrease of leachable aluminum between 70% and 77%. Cell 1B experienced a decrease of leachable aluminum of 54%.

In 2016 there was an overall increase in leachable copper from a depth of 10-20 cm: cell 1A increased by 25%, cell 1B increased by 32%, cell 2A increased by 46%, and cell 2B increased by 7% (Table 8). In 2016 there was a decrease in leachable aluminum from 28% to 58% at a depth of 10-20 cm in cell 1A, 1B, and 2A. Cell 2B increased in leachable aluminum by 49% (appendix D).

Analysis of the soils by ICP-MS with sequentially extracted acid analysis shows that despite elevated initial leachable copper concentrations, the soils have become more stable (less leachable) over time in the wetland setting as the soils have aged (Figure 8). This beneficial aging of soils to a less soluble mineralized form of sulphide is expected for this type of treatment wetland design. It should be noted that due to the starting soil substrate containing leachable copper and other metals such as aluminum (Appendix B) that these elements are leaching from the substrate into the water, putting additional treatment demands on the systems.

The evapotranspiration study also confirmed that due to copper, metals, and metalloids in the initial substrates placed in the soils, constituents continue to leach from the substrate as these constituents shift from oxidized fractions 2 and 3 to reduced fraction 4 (Appendix B). Copper concentrations in the CWTS increased on average from 58 µg/L in in 2016 prior to evaporation study, to an average 80 µg/L during the evapotranspiration study. Additionally, aluminum concentrations in the CWTS in 2016 increased on average from 10.9 µg/L prior to evaporation study, to an average 32 µg/L during the evapotranspiration study. Molybdenum, selenium, and zinc all decreased in concentrations during the evapotranspiration study compared to the average concentration prior to the study in 2016. Molybdenum decreased from ~7.5 µg/L to ~6.0 µg/L, selenium decreased from ~3.3 µg/L to ~1.3 µg/L, and zinc decreased from ~52 µg/L to ~28 µg/L. Cadmium concentrations were so low that there was not a notable change in concentration during the evaporation study.

It is noted that the increase in copper concentration during the evapotranspiration trial is not representative of what would occur during periods with no flow in a full-scale system, where soils with minimal leachable copper are used and copper is deposited in sulphide form (fraction 4) by the biogeochemical activity of the CWTS. Rather, enhanced removal of constituents such as copper and selenium is expected during these periods as the system becomes more reducing with no flow. In the case of the demonstration-scale wetland, due to the elements switching from oxidized to reduced forms in the soils, the increased concentration of some constituents is seen despite a substantial decrease of sulphate. That is, there are more metals switching from oxidized to reduced forms, than sulphides being produced during this time based on available organic carbon to produce the sulphides (sulphate decreased from ~134 mg/L to ~104 mg/L). This was one of the pieces of data that prompted the addition of carbon sources to the wetland to initiate commissioning-B and address these remaining oxidized metals in the CWTS. If metals continue to leach in 2017, flows could be stopped temporarily to promote this shift from oxidized to reduced forms, with the additional carbon sources of straw and wood chips now available for the bacteria to use to produce more sulphides for metals treatment.

For the CWTS to treat the copper from the soils, additional organic carbon is needed beyond that necessary for the waters alone. Organic carbon is contributed to the CWTS through decomposition of plant material as the wetland matures. However, in 2015 the CWTS did not host enough plants that would decompose and contribute organic carbon in 2016 to the wetland, as plants were still establishing. Therefore, in 2016 additional carbon was added to the wetland in the form of woodchips and straw (Section 5.1). Organic carbon is used as food by the microbes, which in turn produce sulphides which drive metals treatment through mineralization by coupled biogeochemical processes. Presently, as there are excess copper forms transitioning from oxidized to reduced in the soils, these will bind to the sulphides, transforming them to the sulphide bound fraction 4 (Table 7). This in turn impacts the ability to treat water, as it uses the available electrons and results in insufficient sulphides to treat metals in the water. When the mineral fractions of the soils have completed transforming the oxidized copper in the soils to a sulphide bound fraction 4 mineral, the soils will no longer be sulphide hungry, allowing for the soil-produced sulphides to be utilized for metals treatment in the water. This progress can be followed by the Tessier extractions, and also by the appearance of acid volatile sulphides (AVS) in the soils. AVS was tested for in 2016 prior to adding the straw and wood chips and was non-detectable (hence, the decision to add the straw and wood chips). The appearance of measurable AVS over time will indicate that the microbes in the soil have worked through the additional copper and the wetland should start performing the way it should have had soils with high leachable copper not been used. In 2017 an AVS soil test should be conducted to determine if the soils have completed the transition to fraction 4.

It is recommended that for construction of the full-scale systems, soils with low total and leachable copper concentrations should be used. Additionally, to supplement carbon until the vegetation has fully established, solid phase organic carbon (e.g., straw, woodchips) should be mixed into the soils. After plants have begun to grow, but are not yet fully mature and able to provide enough annual carbon to the wetland through their growth, solid phase organic carbon can also be added into the wetland in the same way as was done in the demonstration-scale CWTS in 2017.

Table 8 – Total and leachable soil copper concentrations in first year of operations.

CWTS Cell	Sampling Date	Days in Operation	Sample Depth (cm)	Total Cu (mg/kg)	SPLP Cu (mg/L)	Leachable Cu (mg/kg)
1A	28-Aug-14	1	Initial Composite	-	0.298 (0.148-0.608)	-
	19-Jun-15	58	0-5	960	0.055[1]	-
	16-Aug-15	116	10-20	950	0.187	-
	18-Sep-15	149	10-20	1300	0.049	-
	29-Sep-15	160	10-20	910	0.069	-
	15-Jun-16	194	0-10	1440	-	0.953
	15-Jun-16	194	10-20	1210	-	0.66
	8-Jul-16	217	0-10	1430	-	0.603
	8-Jul-16	217	10-20	1730	-	0.832
	7-Sep-16	247	0-10	1290	-	0.449
1B	28-Aug-14	1	Initial Composite	-	0.298 (0.148-0.608)	-
	19-Jun-15	58	0-5	1400	0.033[1]	-
	16-Aug-15	116	10-20	1400	0.209	-
	18-Sep-15	149	10-20	830	0.065	-
	29-Sep-15	160	10-20	880	0.059	-
	15-Jun-16	194	0-10	1130	-	1.01
	15-Jun-16	194	10-20	1240	-	0.822
	8-Jul-16	217	0-10	1250	-	1.11
	8-Jul-16	217	10-20	1620	-	1.21
	7-Sep-16	247	0-10	1190	-	0.197
2A	28-Aug-14	1	Initial Composite	-	0.298 (0.148-0.608)	-
	19-Jun-15	58	0-5	1175	0.037[1]	-
	16-Aug-15	116	10-20	660	0.139	-
	18-Sep-15	149	10-20	880	0.081	-
	29-Sep-15	160	10-20	1000	0.073	-
	15-Jun-16	194	0-10	1100	-	0.823
	15-Jun-16	194	10-20	1280	-	0.900
	8-Jul-16	217	0-10	1450	-	0.963
	8-Jul-16	217	10-20	1150	-	1.68
	7-Sep-16	247	0-10	1290		0.838

Table 8 – Continued.

CWTS Cell	Sampling Date	Days in Operation	Sample Depth (cm)	Total Cu (mg/kg)	SPLP Cu (mg/L)	Leachable Cu (mg/kg)
2B	28-Aug-14	1	Initial Composite	-	0.298 (0.148-0.608)	-
	19-Jun-15	58	0-5	1100	0.039[1]	-
	16-Aug-15	116	10-20	1000	0.201	-
	18-Sep-15	149	10-20	830	0.078	-
	29-Sep-15	160	10-20	540	0.059	-
	15-Jun-16	194	0-10	1040	-	0.771
	15-Jun-16	194	10-20	1310	-	1.16
	8-Jul-16	217	0-10	1070	-	1.38
	8-Jul-16	217	10-20	1180	-	1.25
	7-Sep-16	247	0-10	1520	-	0.123

[1] Samples collected in June 2015 were at a shallow depth (0-5 cm) and copper content had therefore likely already been removed by washing from the faster flows of the CWTS system. The aging of the deeper (10-20 cm) soil copper concentrations over time to a less soluble form is shown in Figure 8.
The blue shading indicated samples that were taken from a shallower depth.

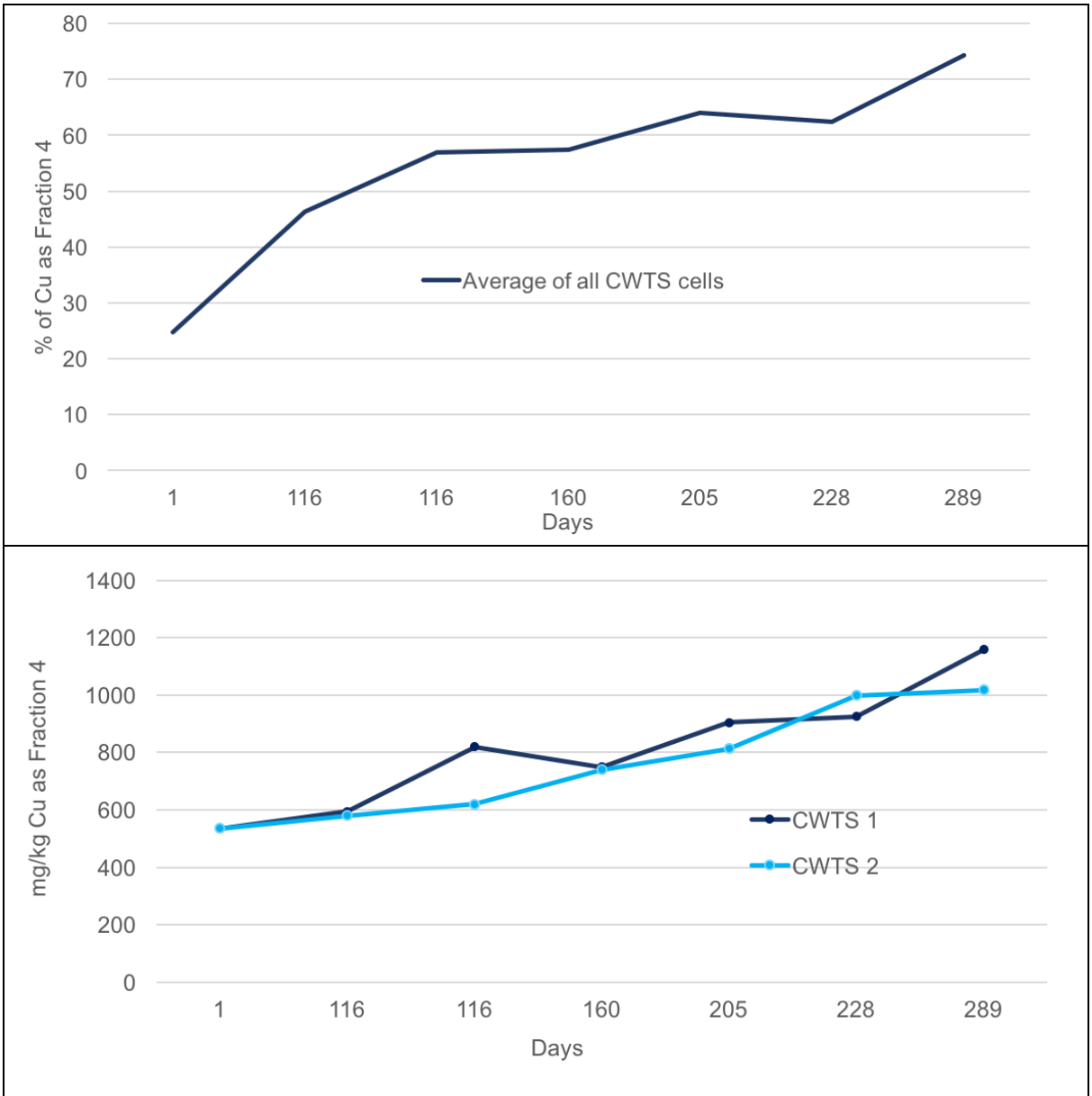


Figure 8 – Copper as sequential leach fraction 4 over time.
Copper in the form of the targeted Fraction 4 (sulphide mineral form and bound to organics), increased in the CWTS over time as the soils matured. Additional information in Appendix B.

6.2.Water

6.2.1. Pilot-scale testing performance

Pilot-scale testing was performed with water that mimicked the worst-case long-term post-closure water chemistry (mean influent concentrations of 146 µg/L copper and 10.2 µg/L selenium). In contrast, the on-site demonstration-scale system uses water from the sump downslope of the dry stack tailings facility and the MVFE (W36 or W62) and therefore

has inflow concentrations that are currently occurring during operations (mean influent concentrations of 58 µg/L copper and 6 µg/L selenium). Because of this, a lower percent removal will achieve the same extent of removal and associated outflow concentrations than was required in the pilot-scale systems. For example, the pilot-scale system achieved on average 92% removal of copper (mean outflow of 11.3 µg/L) and 41% removal of selenium (mean outflow of 6 µg/L). To achieve these same outflow concentrations with the demonstration-scale system, 80% removal of copper and minimal (or no) removal of selenium would be needed.

6.2.2. Demonstration-scale metal leaching from soils

The treatment of copper has been improving through commissioning as the system has matured (Figure 16). However, as shown in Figure 10 and Figure 12 several elements are being released through the wetland by the soils, such as aluminum, and copper (as discussed in Section 6.1). Because of this, negative percent removals of copper have been observed, especially in early operations (Appendix B). It was initially theorized that this was occurring when the May and June 2015 data reported outflow concentrations higher than the influent. This paired with the soils data (Section 6.1 and Appendix A) led to the design of a new sampling scheme to determine whether detectable levels of metals were being released from the soils within the wetland (2015) and to monitor this (2016). For two timepoints in 2015 (August 15, 2015 and September 17, 2015) and 2 timepoints in 2016 (June 14, 2016 and September 7, 2016), water was sampled not only at the feed and the outflow of each cell, but also through the cell. Beginning at the outflow of cell B and working upwards towards the feed, samples were taken at the B cell outflow, B cell mid-point, and B cell inflow, and then the same 3 points for the A cell (Figure 9). In each case, the sample was taken from within a reaching distance of the side shore to ensure that sediments were not suspended in sampling.

Since the beginning of operations of the demonstration-scale CWTS, significant concentrations of copper have been leaching from the wetlands soils of all cells (Figure 12). Moreover, metal types and concentrations (e.g., Aluminum, Appendix B) that could not be accounted for by the influent water chemistry were elevated at random points within the wetland. This suggested that because of metals leaching from the soils, the treatment occurring within the wetlands was far greater than what was being observed by simply measuring the inflow and outflow points (Figure 16). For example, in August of 2015, influent copper concentrations were measured as 50 µg/L, but within the first meter of cell 1A increased to 100 µg/L, and by the outflow of cell 1A, down to 42 µg/L (58 µg/L decrease), they then increased to 70 µg/L at the beginning of cell B, and again were treated to a final outflow concentration of 40 µg/L (30 µg/L decrease) for a total of at least 88 µg/L removed by Series 1 of the treatment wetland. This suggested the wetland is achieving much greater copper treatment (88 µg/L removed), than would be suggested by only measuring the inflow and outflow of the system (suggests 10 µg/L removed). Similar leaching was observed through the wetland during testing in September 2015.

When the flow through the wetland was tested for metals leaching from soils in September 2016, influent copper concentrations were measured as 51 µg/L, but within the first meter of cell 1A increased to 74 µg/L, and by the midpoint of cell 1A they decreased to 36 µg/L and at

the outflow of cell 1A, increased to 40 µg/L (i.e., a net 39 µg/L decrease, although only appearing as 14 µg/L compared to inflow). Copper concentrations were 34 µg/L at the beginning of cell B, and again were treated to a final outflow concentration of 28 µg/L (6 µg/L decrease) for a total of at least 45 µg/L removed by cells A and B of Series 1 of the treatment wetland. This suggests the wetland is achieving much greater copper treatment (45 µg/L removed), than would be suggested by only measuring the inflow and outflow of the system (suggests 23 µg/L removed). Copper continues to leach from the wetland soils, however the amount of leaching has decreased in 2016, with an 80% decrease in leachable copper from the soil in B cells, and 50% decrease in leachable copper in cell 1A, and cell 2A has become more stable with little change in leachable copper through 2016 (Figure 12).

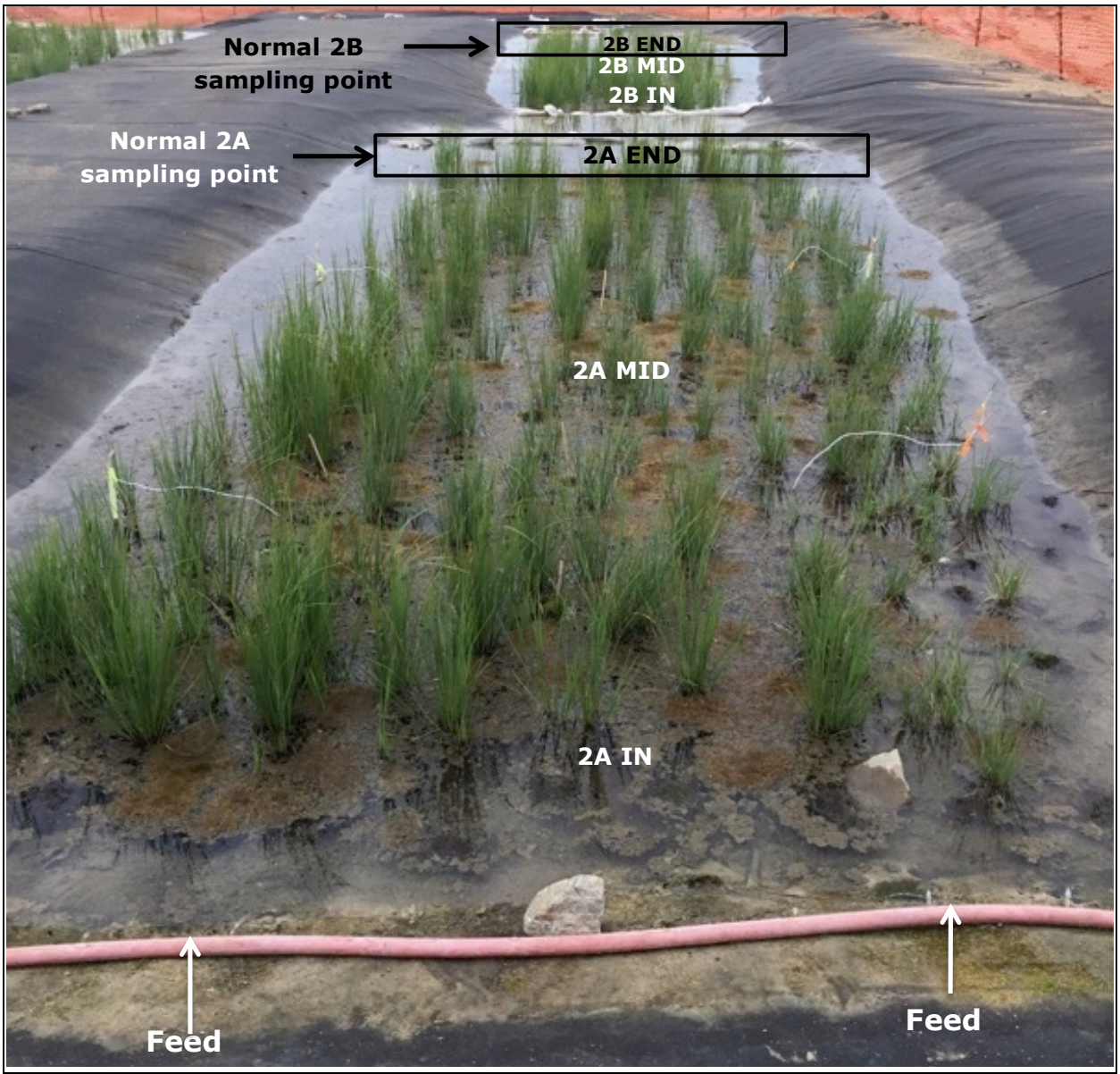


Figure 9 – Sampling points through the CWTS (Photo taken August 16, 2015).

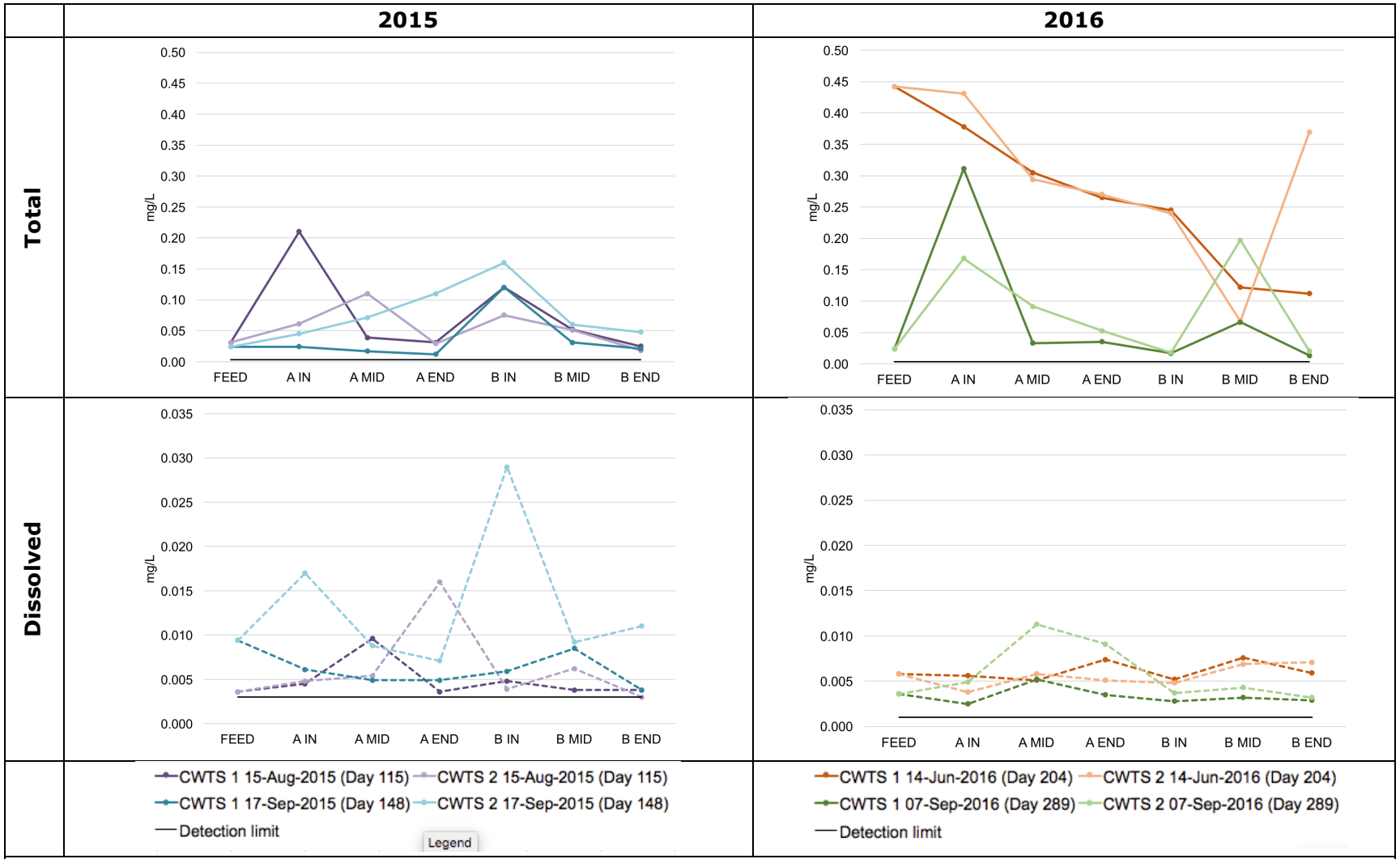


Figure 10 – Aluminum concentrations through the CWTS.
Left is 2015, right is 2016, top graphs are total concentrations and bottom graphs are dissolved concentrations. Data shown for four timepoints, where water was sampled at 7 locations through the flow path of the CWTS to assess for treatment fronts within the wetland, or possible leaching of elements from the soils into the CWTS. Y-axes are set to be the same for total and the same for dissolved. The Maxxam (2015 results) detection limits for aluminum are 0.0030 mg/L. The ALS (2016 results) dissolved aluminum detection limit is 0.0010 mg/L, and the total aluminum detection limit is 0.0030 mg/L.



Figure 11 – Cadmium concentrations through the CWTS.
Left is 2015, right is 2016, top graphs are total concentrations and bottom graphs are dissolved concentrations. Data shown for four timepoints, where water was sampled at 7 locations through the flow path of the CWTS to assess for treatment fronts within the wetland, or possible leaching of elements from the soils into the CWTS. Y-axes are set to be the same for total and the same for dissolved. The Maxxam (2015 results) detection limits for cadmium are 0.020 µg/L. The ALS (2016 results) detection limits are 0.005 µg/L.

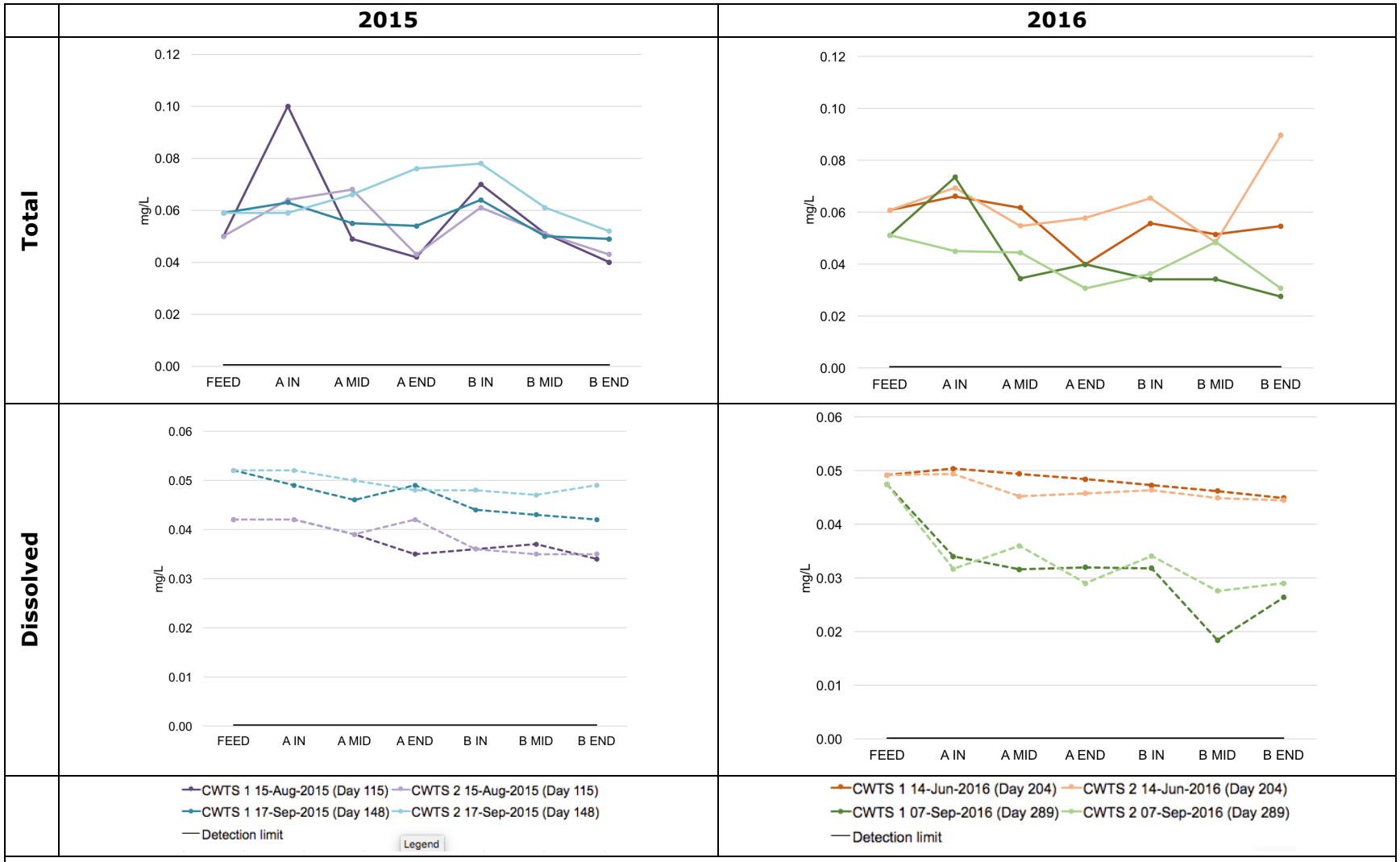


Figure 12 – Copper concentrations through the CWTS.
Left is 2015, right is 2016, top graphs are total concentrations and bottom graphs are dissolved concentrations. Data shown for four timepoints, where water was sampled at 7 locations through the flow path of the CWTS to assess for treatment fronts within the wetland, or possible leaching of elements from the soils into the CWTS. The Maxxam (2015 results) detection limits for copper are 0.0002 mg/L. The ALS (2016 results) detection limits for copper are 0.0005 mg/L.



Figure 13 – Molybdenum concentrations through the CWTS.
Left is 2015, right is 2016, top graphs are total concentrations and bottom graphs are dissolved concentrations. Data shown for four timepoints, where water was sampled at 7 locations through the flow path of the CWTS to assess for treatment fronts within the wetland, or possible leaching of elements from the soils into the CWTS. Y-axes are set to be the same for total and dissolved. The Maxxam (2015 results) detection limits for molybdenum are 0.0002 mg/L. The ALS (2016 results) detection limits are 0.000050 mg/L.

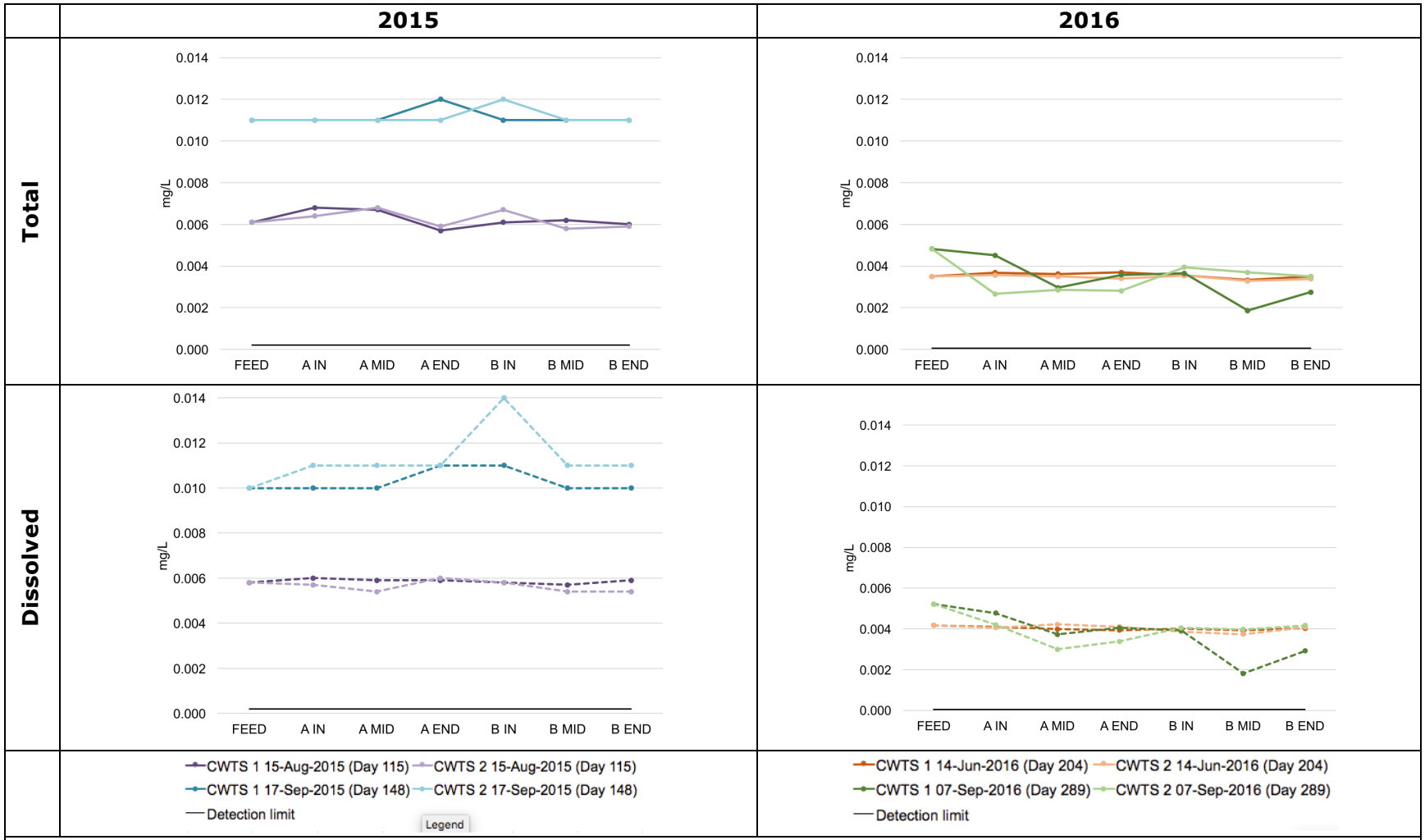


Figure 14 – Selenium concentrations through the CWTS.
Left is 2015, right is 2016, top graphs are total concentrations and bottom graphs are dissolved concentrations. Data shown for four timepoints, where water was sampled at 7 locations through the flow path of the CWTS to assess for treatment fronts within the wetland, or possible leaching of elements from the soils into the CWTS. Y-axes are set to be the same for total and dissolved. The Maxxam (2015 results) detection limit for selenium is 0.0002 mg/L. The ALS (2016 results) detection limit is 0.000050 mg/L.

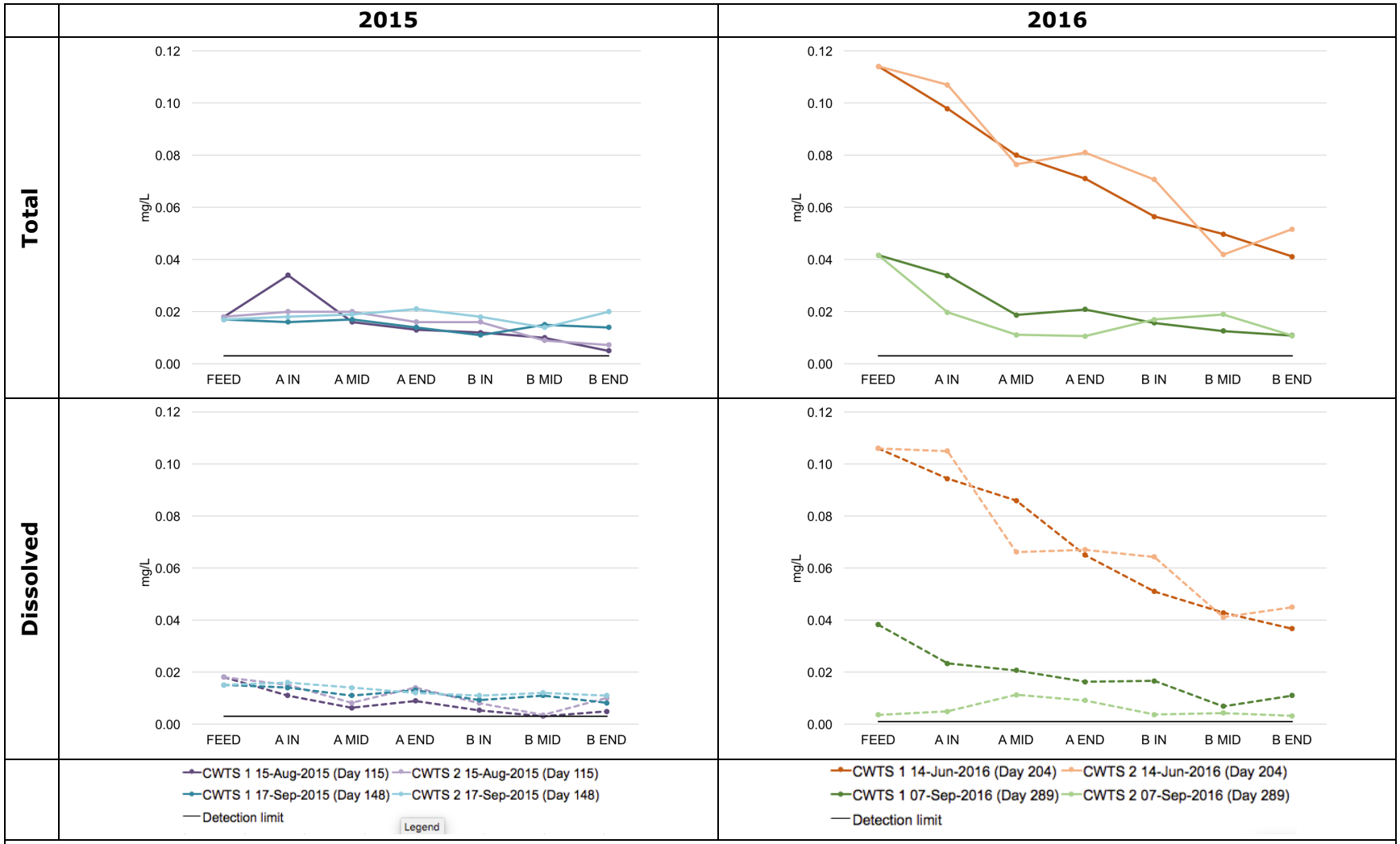


Figure 15 – Zinc concentrations through the CWTS.
Left is 2015, right is 2016, top graphs are total concentrations and bottom graphs are dissolved concentrations. Data shown for four timepoints, where water was sampled at 7 locations through the flow path of the CWTS to assess for treatment fronts within the wetland, or possible leaching of elements from the soils into the CWTS. Y-axis is set to be the same for total and dissolved. The Maxxam (2015 results) detection limit for zinc is 0.0030 mg/L. The ALS (2016 results) detection limit is 0.0010 mg/L.

6.2.3. Demonstration-scale performance during commissioning-A

By the end of 2015 the outflow concentration for dissolved cadmium, was on average 49% lower than influent concentrations, with average influent concentrations of 0.0505 µg/L decreasing to an average outflow concentration of 0.0248 µg/L. During commissioning-A in 2016 treatment decreased to an average removal of 40%, with average influent concentrations of 0.0240 µg/L decreasing to an average outflow concentration of 0.0142 µg/L. Dissolved copper also experienced the same trend with an average outflow concentration 17% lower than influent concentrations at the end of 2015, with average influent concentrations of 54.5 µg/L decreasing to an average outflow concentration of 45.0 µg/L. During commissioning-A in 2016, treatment decreased to an average removal of 8%, with average influent concentrations of 61.6 µg/L decreasing to an average outflow concentration of 56.7 µg/L. Molybdenum experienced no treatment through commissioning-A in 2015 and 2016, with average outflow concentrations being very similar to average influent concentrations. Dissolved selenium also experienced no treatment through commissioning-A in 2015, however by the end of commissioning-A in 2016 dissolved selenium was on average 8% lower than influent concentrations, with average influent concentrations of 3.7 µg/L decreasing to an average outflow concentration of 3.4 µg/L. By the end of 2015 the outflow concentration for dissolved zinc, was on average 42% lower than influent concentrations, with average influent concentrations of 12.3 µg/L decreasing to an average outflow concentration of 6.8 µg/L. During commissioning-A in 2016 treatment increased to an average removal of 47%, with average influent concentrations of 92.8 µg/L decreasing to an average outflow concentration of 49.0 µg/L lower than influent concentrations.

6.2.4. Elevated zinc in influent in 2016

The concertation of zinc in the influent to the CWTS increased significantly in 2016 compared to 2015 concentrations (Table 9, Figure 15, and Figure 20). It is uncertain whether the increase in zinc is a result of the change in sump source water location at the toe of the MVFE, where a metal culvert was installed at the beginning of operations in 2016 (Section 2.4), or if this is a result of seasonal variation.

6.2.5. Demonstration-scale performance during commissioning-B

After commissioning-A was complete at the end of July 2016, cadmium, copper, molybdenum, selenium, and zinc were being treated by the CWTS (Figure 16 to Figure 21 and Appendix B), while other metals/metalloids of potential concern as outlined in the Adaptive Management Plan (AMP) (Minto, 2016) were below the non- degradation central tendency water quality objectives (NDCT-WQO) for the Lower Minto Creek (W2) area.

Dissolved cadmium treatment improved through the commissioning-A period in 2016 to achieve on average 64% removal during the commissioning-B period from July 29 to September 30, 2016, with average influent concentrations of 0.0185 µg/L decreasing to an average outflow concentration of 0.0066 µg/L (Table 9, Figure 16 and Appendix B) Copper, and zinc experienced the same trends with treatment improving through the commissioning-A period in 2016 to achieve on average 37% and 69% removal during the commissioning-B

period from July 29 to September 30, 2016 (Table 9, Figure 17, Figure 20, and Appendix B). During the commissioning-B period, the average influent concentrations for copper (46.1 µg/L) decreased to an average outflow concentration of 28.8 µg/L. During the same time period the average influent concentrations for zinc (37.6 µg/L) decreased to an average outflow concentration of 11.6 µg/L. The treatment of molybdenum and selenium in 2016 is notable as there was minimal treatment in 2015, improving through 2016 to achieve on average 21% and 41% removal during the commissioning-B period from July 29 to September 30, 2016 (Table 9, Figure 18, Figure 19, and Appendix B). This suggests that not only is the wetland maturing as expected, creating reducing sulphide-producing conditions, but that it is already performing far beyond what was anticipated from the design, sequestering orders of magnitude more copper than would be apparent by looking at the influent water alone (i.e., is also removing copper that has leached from the soils over time). All constituents of concern have percent removal rates that continue to trend upward and are becoming more stable as the wetland matures.

Table 9 – Percent removal of constituents in the demonstration-scale CWTS.

Element (µg/L)		Pilot-scale[1]	Demonstration-scale			
		2013	2014	2015[2]	2016 [3]	2016[4]
Cd	In	0.336	NA	0.0505	0.0240	0.0185
	Out	0.027	NA	0.0248	0.0142	0.0066
	%	**92**	NA	**49**	**40**	**64**
Cu	In	146	NA	54.5	61.6	46.1
	Out	11.3	NA	45.0	56.7	28.8
	%	**92**	NA	**17**	**8**	**37**
Mo	In	13.0	NA	11.0	7.6	7.3
	Out	10.4	NA	11.3	7.6	5.7
	%	**20**	NA	**0**	**0**	**21**
Se	In	10.2	NA	11.0	3.7	5.7
	Out	6	NA	11.3	3.4	3.3
	%	**41**	NA	**0**	**8**	**41**
Zn	In	40	NA	12.3	92.8	37.6
	Out	3.2	NA	6.8	49.0	11.6
	%	**92**	NA	**42**	**47**	**69**

[1]The pilot-scale system operated from Nov 4, 2013 to July 16, 2014. The data used in this table is from the low N period in the *carex* and moss system which operated from April 13, 2014 to May 28, 2014.
[2] Values calculated from end last two sampling event in 2015 before the addition of EDTA (September 9 and 17, 2015).
[3] Commissioning-A (May 2, 2016 to July 28, 2016).
[4] Commissioning-B (July 29, 2016 to September 30, 2016).

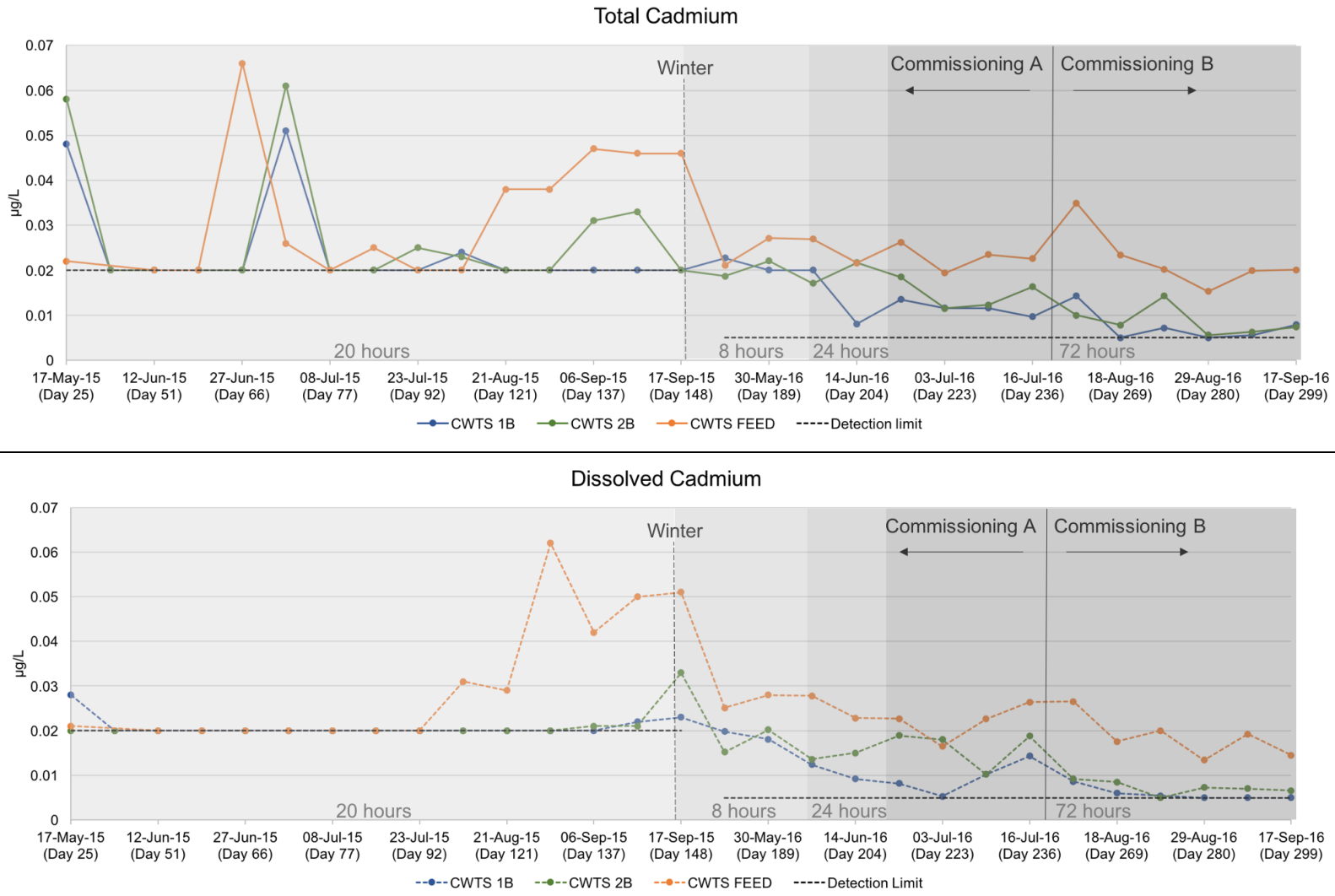
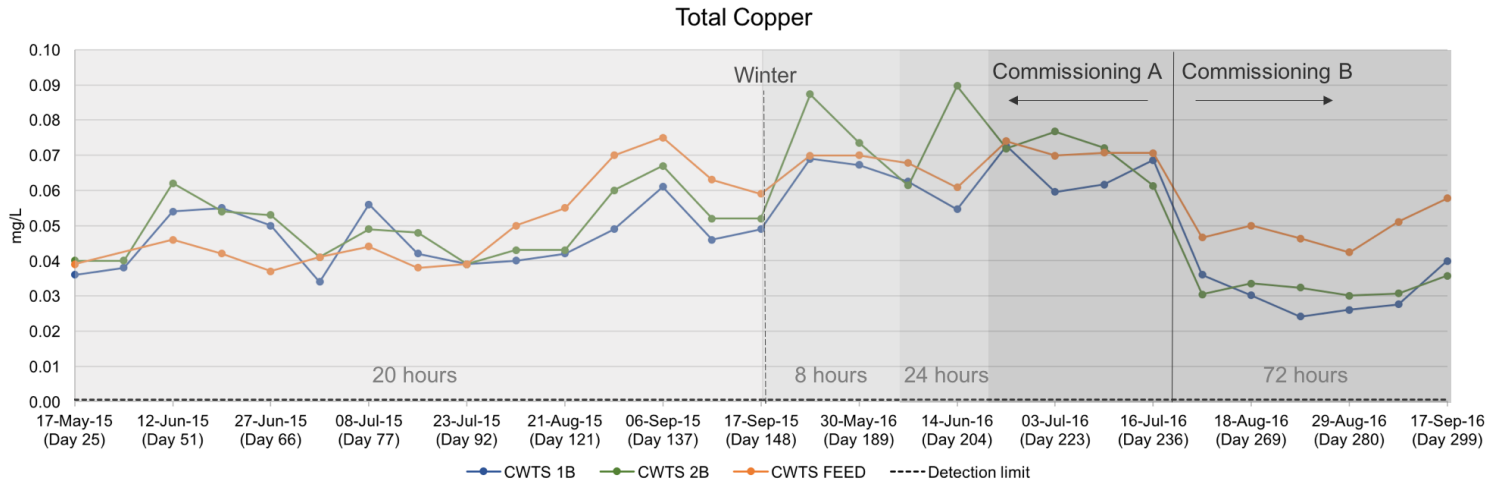


Figure 16 – Cadmium concentrations in the demonstration scale wetland.
The HRT is indicated by grey shading and the duration in hours. The Maxxam (2015 results) measurement uncertainty (MU) for cadmium is +/- 0.020 µg/L. The ALS (2016 results) MU is +/- 0.005 µg/L. Timepoints where the dissolved concentration is higher than the total concentration are within the limit of uncertainty.



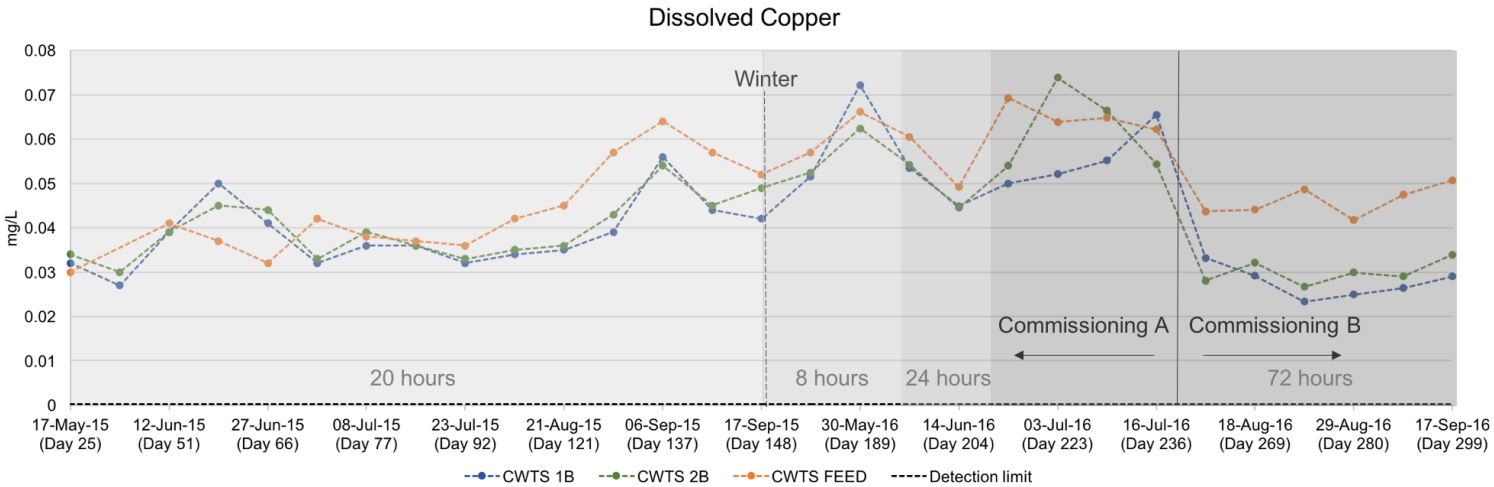


Figure 17 – Copper concentrations in the demonstration scale wetland.
The HRT is indicated by grey shading and the duration in hours. The Maxxam (2015 results) and The ALS (2016 results) measurement uncertainty (MU) for copper is +/- 0.0002 mg/L. Timepoints where the dissolved concentration is higher than the total concentration are within the limit of uncertainty.

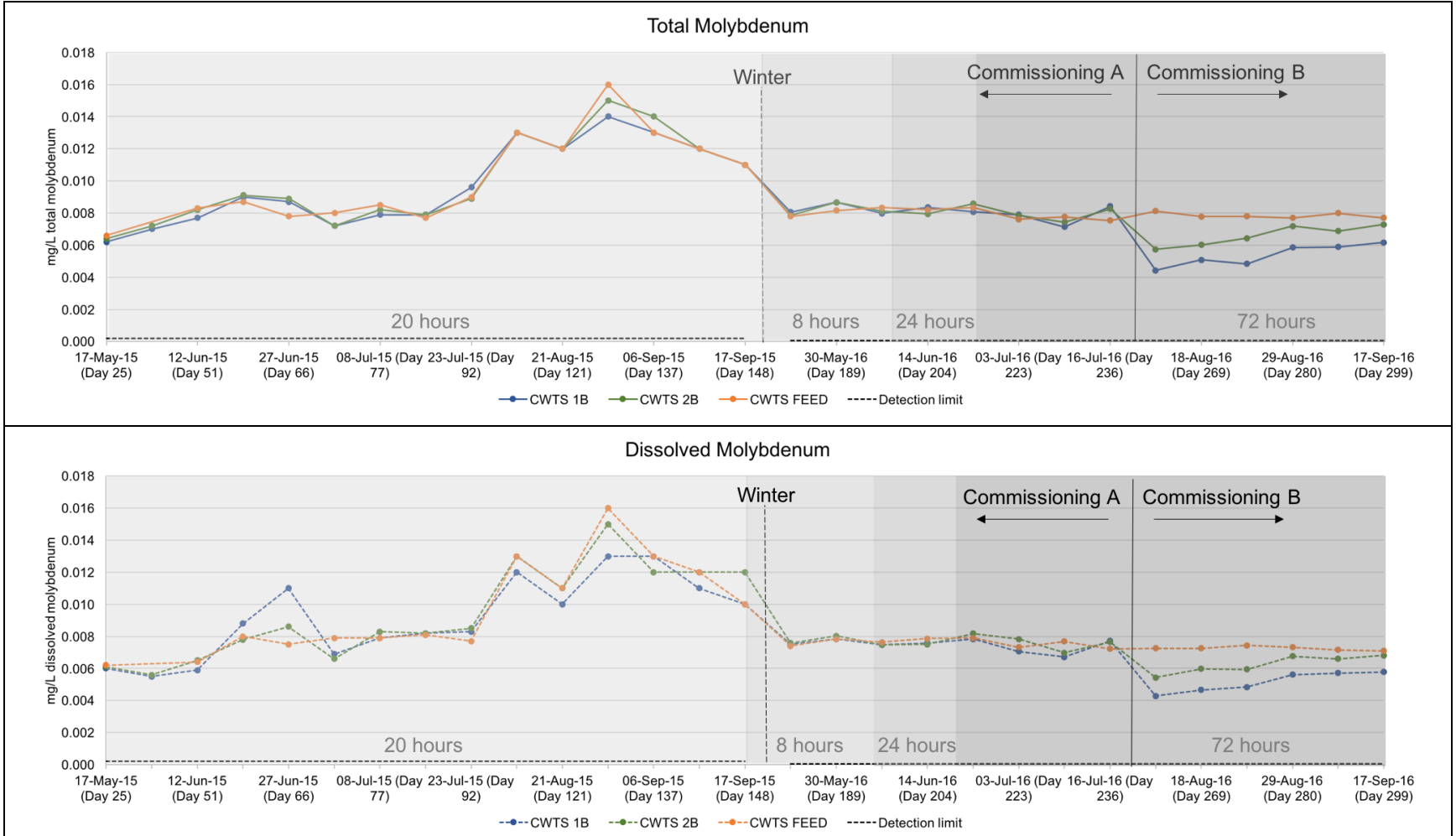
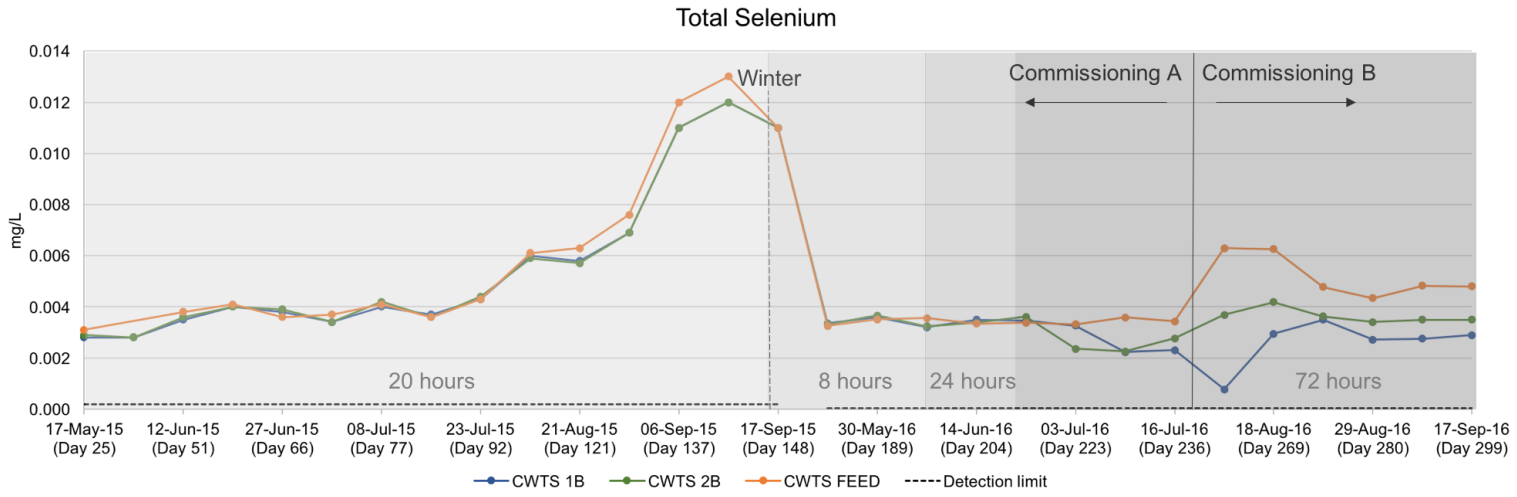


Figure 18 – Molybdenum concentrations in the demonstration scale wetland.
The HRT is indicated by grey shading and the duration in hours. The Maxxam (2015 results) measurement uncertainty (MU) for molybdenum is +/- 0.0002 mg/L. The ALS (2016 results) MU is +/- 0.000050 mg/L. Timepoints where the dissolved concentration is higher than the total concentration are within the limit of uncertainty.



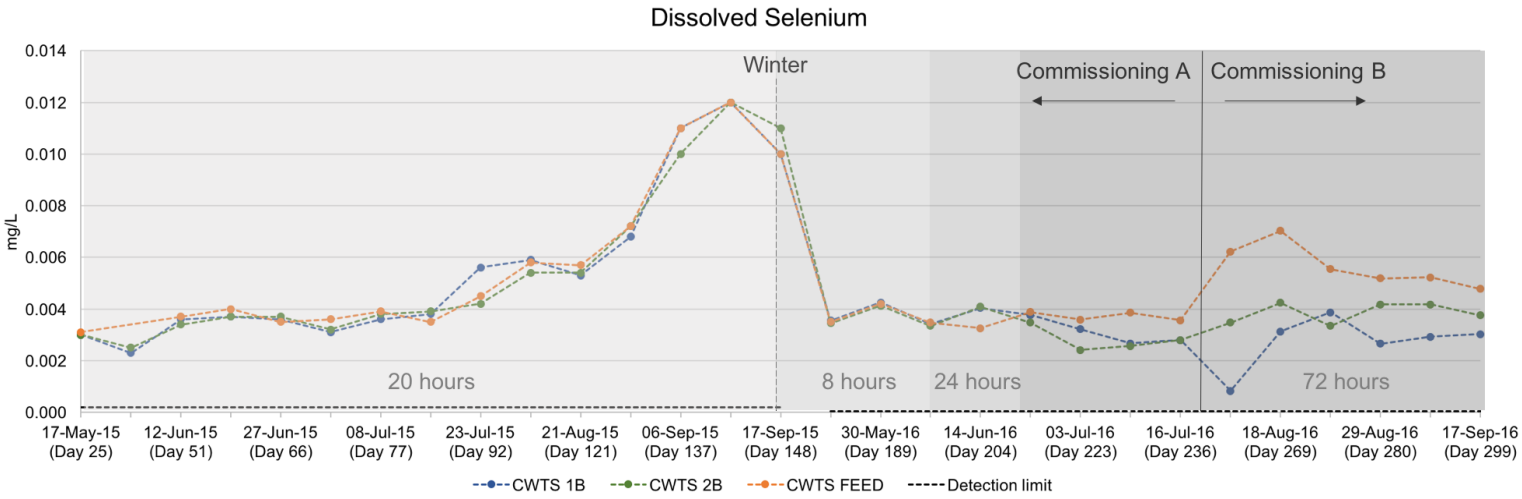


Figure 19 – Selenium concentrations in the demonstration scale wetland.
The HRT is indicated by grey shading and the duration in hours. The Maxxam (2015 results) measurement uncertainty (MU) for selenium is +/- 0.0002 mg/L. The ALS (2016 results) MU is +/- 0.000050 mg/L. Timepoints where the dissolved concentration is higher than the total concentration are within the limit of uncertainty.

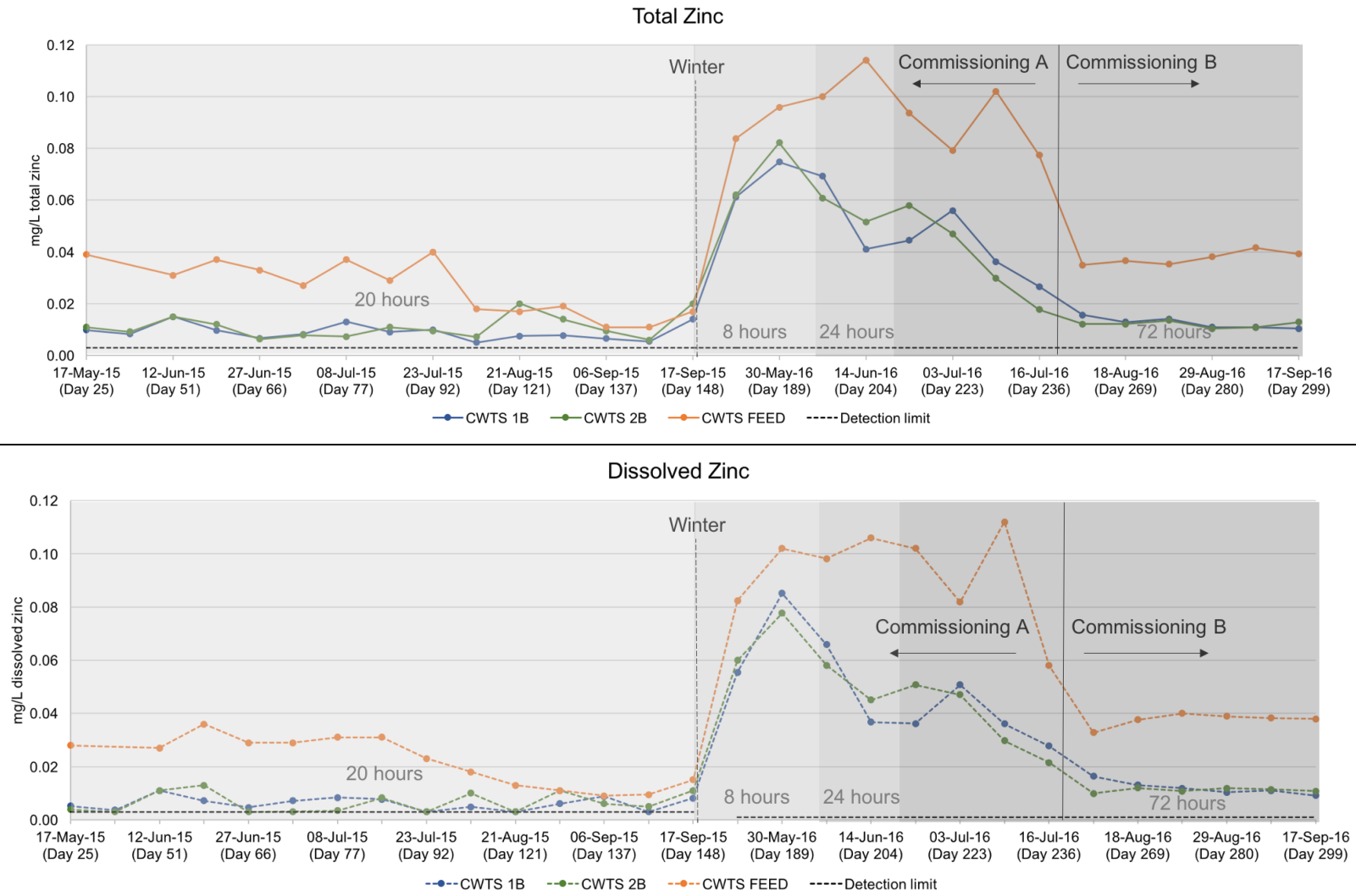


Figure 20 – Zinc concentrations in the demonstration scale wetland.
The HRT is indicated by grey shading and the duration in hours. The Maxxam (2015 results) measurement uncertainty (MU) for zinc is +/- 0.0030 mg/L. The ALS (2016 results) MU is +/- 0.0010 mg/L. Timepoints where the dissolved concentration is higher than the total concentration are within the limit of uncertainty.

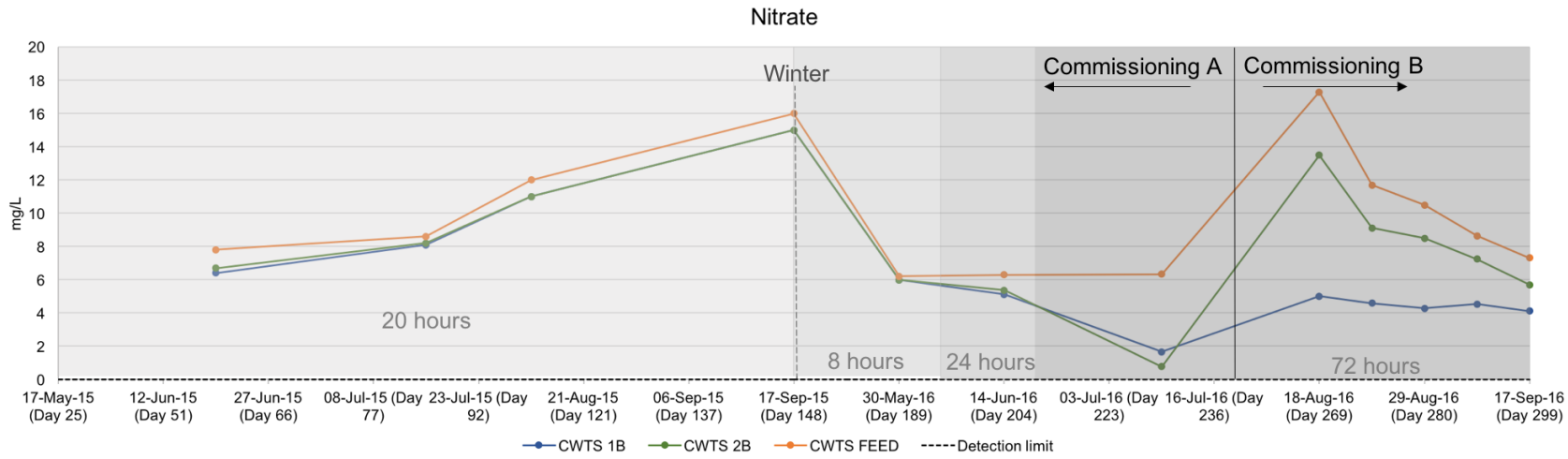


Figure 21 – Nitrite as N (NO_2) and Nitrate as N (NO_3) concentrations in the demonstration scale wetland.
The HRT is indicated by grey shading and the duration in hours. The Maxxam (2015 results) measured uncertainty (MU) for nitrite is +/- 0.010 mg/L. The ALS (2016 results) MU is +/- 0.0010 mg/L. The Maxxam (2015 results) measured uncertainty for nitrate is +/- 0.010 mg/L. The ALS (2016 results) MU is +/- 0.0050 mg/L.

6.3.Chelated iron (fertilizer micronutrient) treatment

Because elevated metals were present in the soils used for construction, there was an opportunity to assess whether non-invasive treatments could be used to further leach the metals and improve soil quality. A chelated iron (Fe-EDTA) product was added to the Series 2 wetland on September 17, 2015. The Fe-EDTA treatment was performed on Series 2, with Series 1 remaining untreated for comparison purposes. Increases in copper concentrations were observed in the outflow from both cells in CWTS Series 2 after the addition of Fe-EDTA. In the 10 days following the addition of Fe-EDTA, approximately 75 g of copper was removed from the wetland Series 2. This is approximately 1.25 g of copper per m^2 of wetland. Elevated flow rates were selected for CWTS start up in 2016 to flush copper leaching from soil (associated with 2015 EDTA test).

7. Hydraulic retention time tracer study

A tracer study was conducted from June 13-15, 2016 on cell 1A of the CWTS with a flow rate of 2.43 GPM (13.25 m^3/day). The purpose of the tracer study was to determine the actual hydraulic retention time, and the flow path of water through the CWTS. Salt (NaCl) was used as the tracer in this study. By adding a salt solution at the inflow of the wetland, and placing a YSI meter at the outflow of the wetland, it can be determined how long it takes for the salt to pass through the wetland by monitoring the specific conductance (SPC) in situ (as salt raises the SPC). The HRT of the CWTS can be determined through a calculation using the volume and concentration of salt solution added at the inflow of the wetland and the time over which the specific conductance remains elevated above background levels at the outflow.

The HRT tracer study used 1.87 kg of salt dissolved into 480L of water. The 480 L of water was gradually removed from the end of the CWTS, 20 liters at a time, dosed with salt, and added at the front of the CWTS. This method took 50 minutes to dose in the 480 liters of salt solution into the CWTS and was selected as it did not affect the total volume of water in the CWTS during the HRT tracer study. The volume of water and weight of salt was calculated based on the amount required to increase the specific conductance (SPC) of cell 1A to at least twice its background level (i.e., from 850 µs/cm, to 1700 µs/cm) (Oakton Instruments, 1997). The volume of water used was then selected to ensure the concentration of the salt solution was sufficiently diluted and would mix with the water of the CWTS (i.e., wasn't so dense that it would stratify and sink to the bottom of the CWTS).

In situ monitoring of the HRT tracer study was carried out from June 13, 2016 at 14:52 until June 15 at 12:12. This time period was selected because Contango was on site for 2 days to conduct the HRT tracer study and to collect data. Also the battery life of the YSI unit was approximately 2 days, and at the time, it was not feasible to implement dosing pumps and in situ monitoring (data logging) for a longer period of time. The YSI unit was placed in the wetland for approximately 20 mins prior to starting the HRT tracer study to record background levels before dosing of the salt solution into the CWTS. The background SPC prior to commencing the HRT tracer study was approximately 900 µS/cm. During the tracer study, there were two tracer breakthroughs. The first breakthrough was at 1.03 days (24.72 hours),

with peak SPC at 2228 µS/cm. The second breakthrough was at 0.75 days (18 hours), with a peak SPC of 1928 µS/cm. Based on the results of the two breakthroughs in the HRT tracer study, it is possible that there are two flow paths through the wetland, however, this may also be due to how the salt solution was added to the CWTS.

In 2017 it is recommended that another 1 m^3 tank be added to the CWTS area, and another HRT tracer study be conducted. The second $1m^3$ tank could provide a way to mix a lower concentration salt solution, and dose it into the CWTS over a longer period at a constant flow using a metering pump. This would allow the flow rate to be calculated for the duration of time the salt solution is dosed into the CWTS. To conduct the HRT tracer study over a longer period, the YSI unit would need to be connected to a power source due to a limited battery life. By conducting the tracer study in 2017 as described above it is anticipated that only one breakthrough would occur, providing more accurate results.

The data from the 2016 HRT tracer study suggests that the HRT of cell 1A is 0.89 days (21.36 hours), the average of the two breakthroughs. It is possible that not all the tracer salt had yet flowed through the wetland, as SPC had not yet reached original background levels, and so the average HRT may be longer slightly than the 0.89 days observed here. By comparison, the calculated HRT was 0.53 days (12.82 hours), therefore the actual HRT was approximately 40% slower than the calculated HRT. Most of this difference in calculated vs tracer study HRT is owing to the exclusion of pore water involvement in the calculated HRT (because of the soil type used, it was an unknown factor that could not be inferred from the off-site pilot-scale trials). However, part of this difference can also be attributed to loss of water through evapotranspiration (Section 9) which results in a lower flow and therefore longer HRT.

Based on the results of the HRT tracer study, the depth of cell 1A including pore water involvement was able to be calculated following Equation 1. The actual HRT of 0.89 days (21.36 hours) was used to solve Equation 1, resulting in a total depth of 0.35 meters. The measured depth of water in cell 1A was then subtracted from total depth (0.35 meters – 0.16 meters), to obtain a pore water involvement of 0.19 meters. The measured depth of cell 1A was obtained from the depth sticks that were installed in the CWTS in June 2016 (Figure 23).

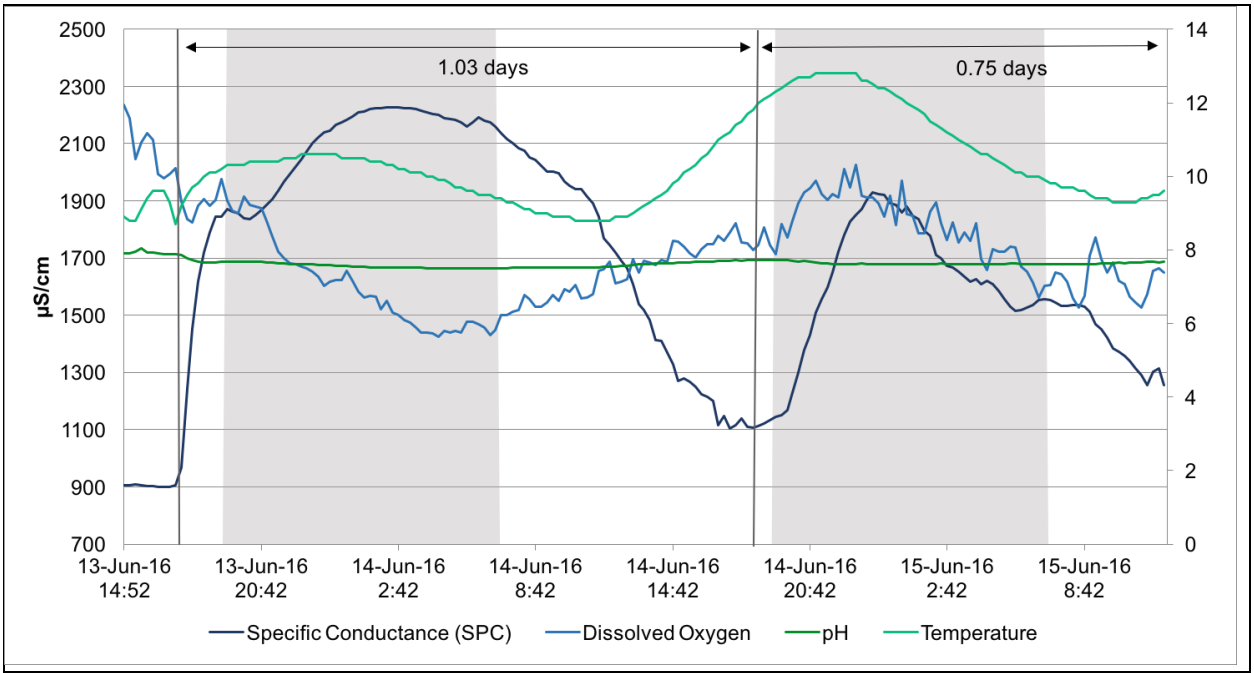


Figure 22 – Tracer study results.
SPC (µS/cm) on primary y-axis. Dissolved oxygen (mg/L), pH, and temperature (°C) on secondary y-axis. The grey shading indicated a 12-hour night period from 7am to 7 pm. The vertical black lines indicate the beginning and end of the tracer breakthrough.

$$\text{Depth} = \left({}^{Q}/_{length * width}\right) * \text{HRT}$$

Equation 1 – Equation for calculation of the CWTS depth including pore water involvement.
Q is the flow rate.

8. Calculations of rate coefficients (*k*)

8.1. Rate coefficients

An important factor for CWTS design is the rate of treatment, also known as the treatment (or removal) rate coefficient (*k*). The rate coefficient is based on the treatability of a specific compound and the hydraulic retention time of the system, both of which are site-specific based on water chemistry, wetland designs, and characteristics of the system. A treatment rate coefficient (*k*) was applied for cadmium, copper, molybdenum, selenium and zinc, to develop a conceptual size for the Minto CWTSs and determine which elements and load sources were key for treatment (Table 10). Based on pilot-scale testing for the Minto CWTS, in this site-specific situation, the treatment rate coefficient (*k*) for selenium follows a zero-order reaction kinetic, while the rate coefficients for cadmium, copper, molybdenum, and zinc follow first-order kinetics.

In Equations 2-3, *Cf* is final concentration, *Ci* is initial concentration, and *HRT* is *the* hydraulic retention (Equation 4) time calculated using the CWTS depth result of the tracer study (Equation 1). The HRT was calculated using the volume of each CWTS series including the pore water involvement calculated from the results of the tracer study (Equation 1), and the flow rate from during the time period that *k* was calculated for in 2016 (from August 10 – September 17, 2016) to obtain the HRT.

Using the removal rate coefficients (*k*) in Table 10 and Equations 2-4, parameters can be rearranged to solve for those of interest, such as the volume needed, that in turn determines the area of wetland required which is dependent upon the design. Analytical results from August 10 – September 17, 2016 were chosen to calculate *k* because this is the period after the final operational adjustment were completed in 2016 (addition of organics) and flow rates were stable and at targeted values. The treatment rate coefficients applied here are intended to be a conservative estimate for conceptual sizing purposes, and will need to be refined through further demonstration-scale (on site) testing.

$$k = \frac{-\ln\left(\frac{Cf}{Ci}\right)}{\mathrm{HRT}}$$

Equation 2 – Equation for calculation of first-order removal rate coefficient.
k is the removal rate coefficient, *Cf* is the final concentration, *Ci* is the initial concentration, and HRT is the hydraulic retention time.

$$k = \frac{(Ci - Cf)}{\mathrm{HRT}}$$

Equation 3 – Equation for calculation of zero-order removal rate coefficient.
k is the removal rate coefficient, *Cf* is the final concentration, *Ci* is the initial concentration, and HRT is the hydraulic retention time.

$$HRT = {}^{V}/_{Q}$$

Equation 4 – Equation for calculation of hydraulic retention time.
HRT is the hydraulic retention time, V is the volume of the wetland, and Q is the flow rate.

Table 10 – Rate coefficients (*k*) for elements in exceedance of proposed WQO's[1].

Element[2]	Pilot *k**hour^{-1}	Demonstration[6] *k**hour^{-1}				Demonstration[6] *k**day^{1}			
		Series 1		Series 2		Series 1		Series 2	
	Total	Dissolved	Total	Dissolved	Total	Dissolved	Total	Dissolved	Total
First order removal rate coefficients									
Cd	0.0476[4]	0.00817	0.00803	0.00610	0.00640	0.196	0.193	0.147	0.154
Cu[3]	0.0497[5]	0.00365[3]	0.00341[3]	0.00283[3]	0.00276[3]	0.0876[3]	0.0819[3]	0.0680[3]	0.0661[3]
Mo	0.0042[4]	0.00248	0.00273	0.000989	0.00117	0.0594	0.0655	0.0237	0.0280
Zn	0.0477[4]	0.00824	0.00790	0.00799	0.00752	0.198	0.190	0.192	0.181
Nitrate	0.0380[4]	N/A	0.00610	N/A	0.00149	N/A	0.147	N/A	0.0358
Zero order removal rate coefficients									
Se	7	0.0000179	0.0000160	0.0000118	0.0000103	0.000430	0.000385	0.000284	0.000246

[1] Non- degradation central tendency water quality objectives (NDCT-WQO) for the Lower Minto Creek (W2) area from the Adaptive Management Plan (Minto, 2016).
[2]All treatment rate coefficients are for first-order reaction kinetics except for selenium which is a zero-order reaction rate kinetic.
[3] Rate coefficients for copper in demonstration scale system are artificially lower than actual due to leaching of copper into water from soils, which increases outflow copper concentrations and affects the *k* calculation (Section 6.2.2).
[4] Values calculated from data in Table 8 of the pilot-scale report (Contango, November 2014).
[5] Value retrieved from Table 10 of the pilot-scale report (Contango, November 2014).
[6]Analytical results from August 10 – September 17, 2016 were used to calculate *k.*

9. Evapotranspiration study

An evapotranspiration study was conducted in a controlled greenhouse at Contango in April 2016 using the same general summer temperatures and vegetation types as the Minto CWTS. Total evapotranspiration from the system is the combined effects of open water evaporation and plant transpiration (Beebe *et al*, 2014). The purpose of calculating the evapotranspiration of a system is to understand the amount of water lost, which in turn concentrates elements and should be considered in the context of the difference of decrease in outflow concentration (or not) and outflow load reduction. The findings of the controlled environment (off site) evapotranspiration study suggested that on average 15.53 L/m^2/day would be lost from a CWTS design such as that at Minto through evapotranspiration. This would result in an equivalent concentration of elements (the actual percentage concentration being dependent on the amount of water in the system).

An evapotranspiration study was then performed on site July 18-25, 2016. The flow to the CWTS was shut off on July 18, 2016 and depth measuring sticks were installed at each end of the wetland to record the amount the water level decreases due to evapotranspiration (Figure 23). However, this time period experienced rain, and also the Minto weather station was not functional at this time, and therefore the evapotranspiration rates of the on-site wetland could not be determined and it is recommended this study should be repeated in 2017.

Applying the findings of the off-site evapotranspiration study to the demonstration-scale CWTS, series 1 could lose 1,085 L/day and series 2 could lose 903 L/day in dry periods. Using the 3 different HRT under which the demonstration-scale CWTS was operated in 2016, evapotranspiration loss could result in a concentration of elements in the outflow water from 3 – 31 % (Table 11), meaning the actual load removed is this much greater than would be appreciated by looking at the concentrations alone. An on-site evaporation trial in 2017 can determine if these findings are accurate for the Minto CWTS. Evapotranspiration must be accounted for in sizing of a full-scale wetland.

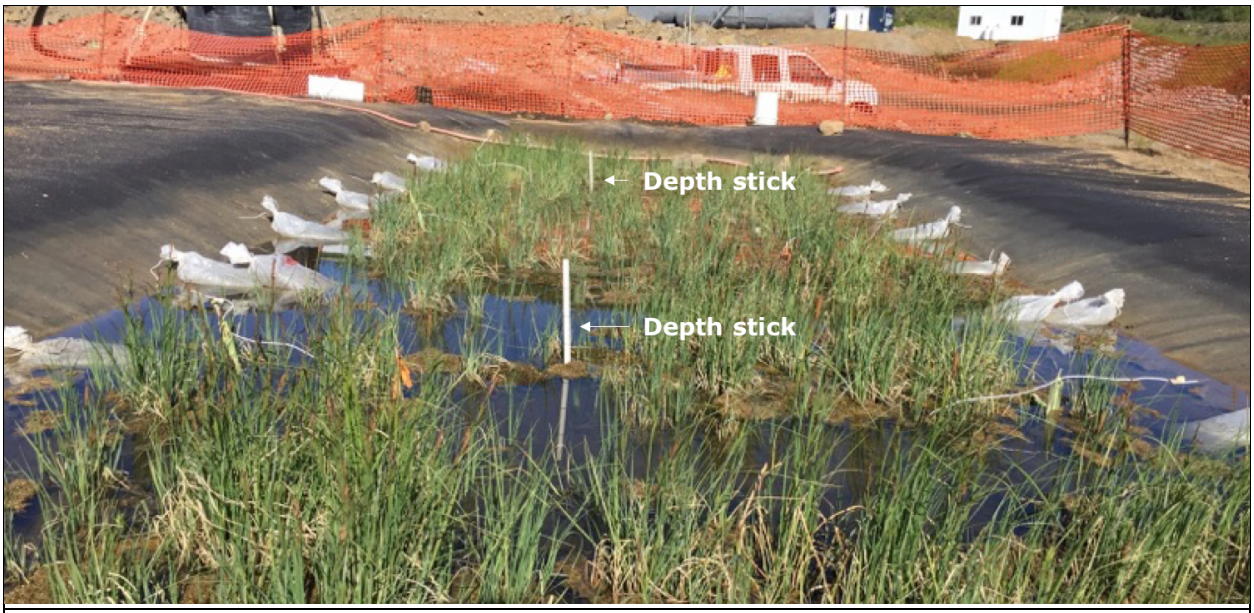


Figure 23 – Depth sticks installed in CWTS to record evapotranspiration.

Table 11 – Potential loss of water through evapotranspiration in the CWTS .

HRT (days)	L/day through series		L/day lost		% Lost /day
	Series 1	Series 2	Series 1	Series 2	
0.3	34,998	28,638	1,085	904	3
1	10,510	8,600			10
3	3,503	2,867			31

10.Vegetation

10.1. Health and establishment

10.1.1. Carex aquatilis

During the commissioning period, plants establish and mature, with density expected to increase over time. When counted at the first site visit in June 2015, >95% of the plants were found to have survived their first winter. This survival rate is impressive considering the late fall planting of August 28, 2014 and suggests the *C. aquatilis* are very robust and reaffirms they are a good candidate for use in the full-scale CWTS at Minto. Additionally, there was a greater than 20% increase in total stems in the short growing season up to June 2015 (as counted by new shoots/runners). By the second site visit in August 2015, the plants were too dense to count stems, and therefore the survival and establishment was considered a success and no longer monitored by counting (Figure 26). By the second site visit in 2016 (July) the CWTS cells hosted a dense monoculture of *C. aquatilis* that continued to thrive until the end of operations in 2016.

Pilot-scale testing suggested that *C. aquatilis* would increase more quickly the number of stems once the maximum plant height was achieved, and that maximum plant height is dependent upon water depth (with deeper water resulting in taller plants). This was also seen in the demonstration-scale wetland on site, with the density of plants filling in greatly through July and August 2015 and throughout 2016 (Figure 26).

Of note, it was observed that during the last site visit of the year by Contango (September 17-18th, 2015), the *C. aquatilis* in the demonstration-scale CWTS were still predominantly green, while other plants in the area had turned brown. This suggests that the CWTSs may have a longer range of activity than natural systems in the area, possibly due to liners and separation from colder ground water.

10.1.1.1. Aphids

Aphids were observed in all CWTS cells in July 2016 (Figure 28), and had originally been noticed in cell 2A in August 2015. Aphids occur naturally in the area and have been observed during site visits and background sampling in the Yukon and Northwest Territories on *Carex* and *Typha*. The effects of aphids on a CWTS are not expected to be pronounced unless it has a large buffer zone between it and other insect habitat (e.g., forest, prairie, etc), which prevents predator insects from also colonizing and keeping the aphids at a lower population density. At Minto, there is a large buffer area (MVFE) around the demonstration-scale CWTS, and as such, the aphids may colonize more robustly than anticipated in the full-scale CWTS at site, which would not have the same buffer zone. The aphids will continue to be watched in 2017, and other background areas (e.g., W10 and W15) compared to see if increases in aphid populations are specific to the CWTS, or general in the area.

10.1.2. Moss

Despite several mosses drying out during the winter of 2015 after planting, there was 100% survival and establishment. However, their growth was much slower than the sedge and therefore they were supplemented with additional aquatic moss (harvested from the same W10 area as the original plants for the CWTS) in June and August 2015 to augment their general abundance and bring the CWTS through commissioning more quickly (Figure 24). Mosses continued to mature and expand in size through 2016. The mature mosses are beginning to show characteristics of the desired coupled transfer (sorption, filtration) and transformation (mineralization, reduction) processes, with the top of the moss growing and producing new biomass (i.e., is green new growth; transfer sites), and the older bottom of the moss turning black (i.e., beginning to decompose producing sulphidic reducing zones; transformation reactions) (Figure 24).

10.2. Metals uptake

Concentrations of various elements in *C. aquatilis* and moss from the demonstration-scale systems were compared to those from pilot-scale testing. It is expected that the concentrations would be higher in the demonstration-scale system as the CWTS is not yet fully functional (i.e., elements are in greater bioavailability when in dissolved form) and also due to the composition of the soils used for construction.

C. aquatilis and moss in the demonstration-scale system had higher copper concentrations than the pilot systems. However, concentration of copper in the *C. aquatilis* at the end of 2015 was not significantly different than that of the plants borrowed from the W10 area for planting of the systems in 2014. Concentration of copper in the *C. aquatilis* in series 1 at the end of 2016 were comparable to the 2015 concentrations, however the concentration of copper in series 2 was higher (Figure 30). This is possibly due to the increased soluble (i.e., bioavailable) copper in series 2 after the addition of Fe-EDTA in September 2015. The moss had higher copper concentrations compared to the initial moss sample in all cells in 2015, and increased to nearly double the 2015 values by the end of 2016 (Figure 30). Mosses remove metals from water through sorption, which is different than the plant uptake that occurs in macrophytes such as *Carex* (and therefore, expected to release copper when exposed to EDTA, rather than have an increased uptake). Additionally, because the mosses were being added through 2015 (from the W10 source), it is expected that mosses would have lower metals concentrations (of any element) in 2015 compared to 2016 as they were not exposed to the CWTS water chemistry for the full year. It should be noted that over time, the mosses convert the sorbed elements to reduced minerals, rendering them lower bioavailability (Section 10.1.2.). As mosses are not a significant food source for any fauna, this is a safe mechanism to transfer elements from water eventually into stable mineral forms in the wetland soils and substrate.

Cadmium concentrations were either below or close to the detection limit in *Carex* in 2015 and 2016 (Figure 29). Additionally, cadmium concentrations in moss and selenium concentrations in both plant types tested were lower in the demonstration-scale (2015 and

2016) compared to pilot-scale. This result was presumably due to lower concentrations of both cadmium and selenium in the water of the demonstration-scale system than pilot scale. In 2016 molybdenum concentrations in *Carex* tissue were higher than the pilot-scale system, but lower than the 2015 CWTS concentrations. In 2016 zinc concentrations in *Carex* tissue has increased compared to the pilot-scale system and the 2015 demonstration-scale system concentrations. This could be attributed to the increase in zinc in the feed water in 2016 (Figure 20). Additionally, in 2016 molybdenum, selenium, and zinc concentrations in moss increased compared to the 2015 concentrations, which suggests that as mosses are continuing to grow and fill in the wetland. Selenium treatment performance is increasing as mosses are a source of sorption, and also harbor a high abundance of selenate-reducing microorganisms (see Section 10.3.2).



Figure 24 – Moss maturation in CWTS.



Figure 25 – Moss top, bottom, and side view June 2016.



Figure 26 – Maturation of the CWTS from construction to year-end 2016.
Top left picture is of cell 1A, top right picture is of cell 2B, bottom left picture is of cell 1B, and the bottom right picture is of cells 2A and 2B. Continued next page

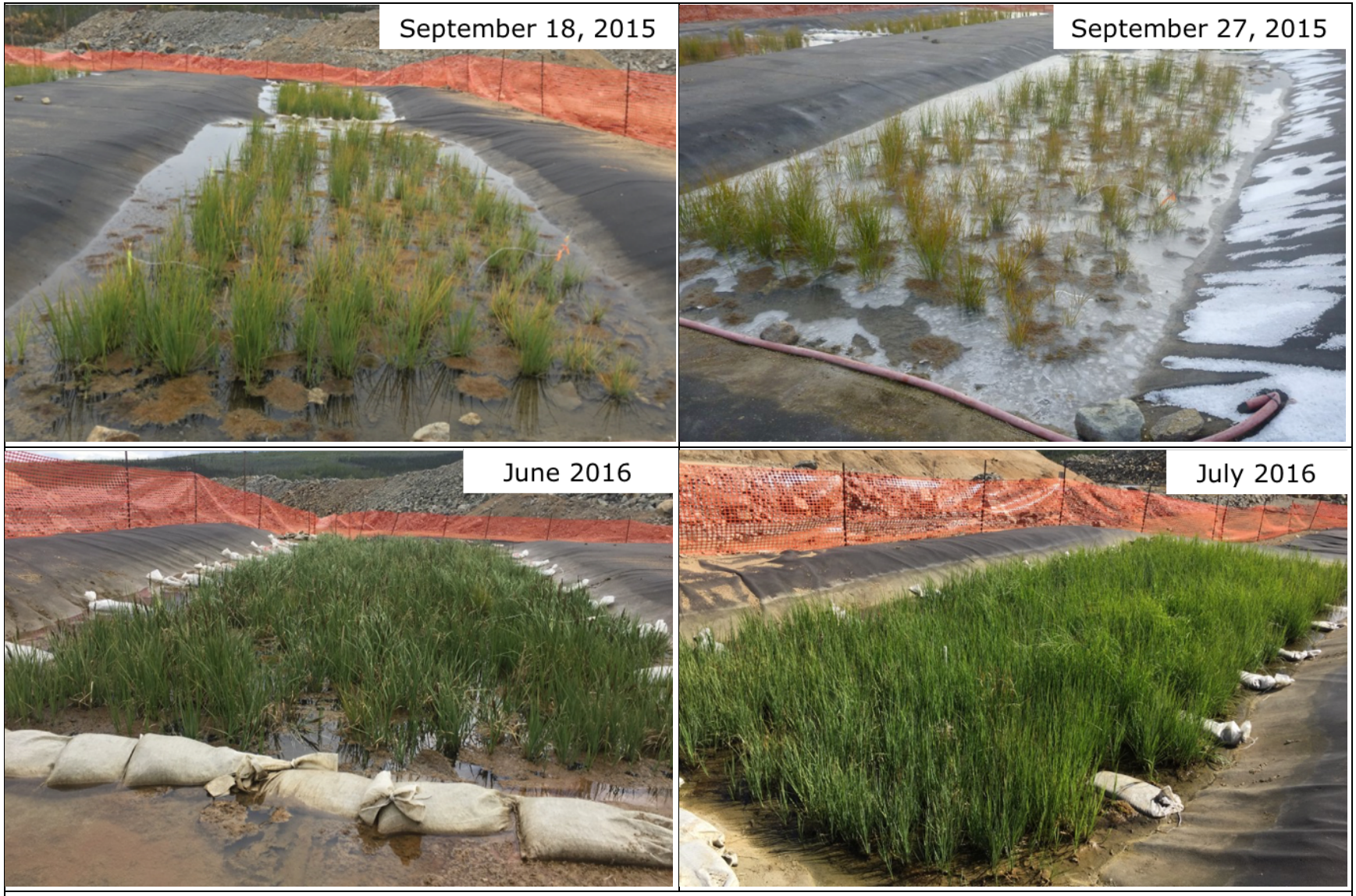


Figure 26 – Continued.
Top pictures are of cell 2A, and bottom picutes are of cell 1B.

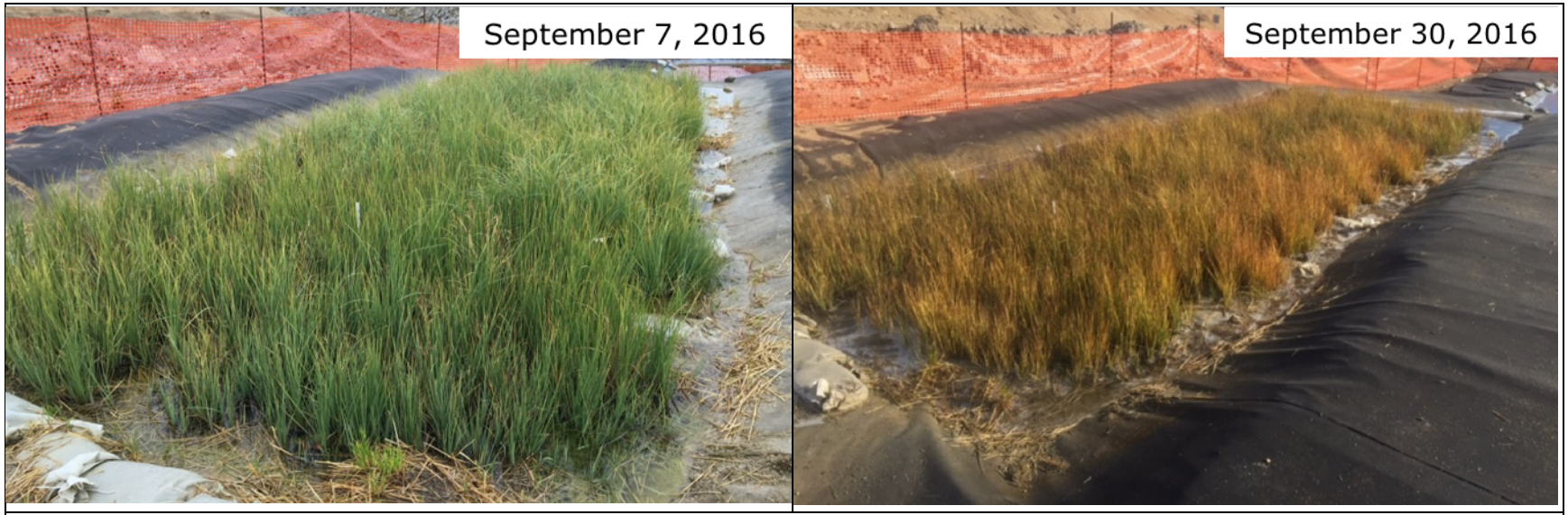


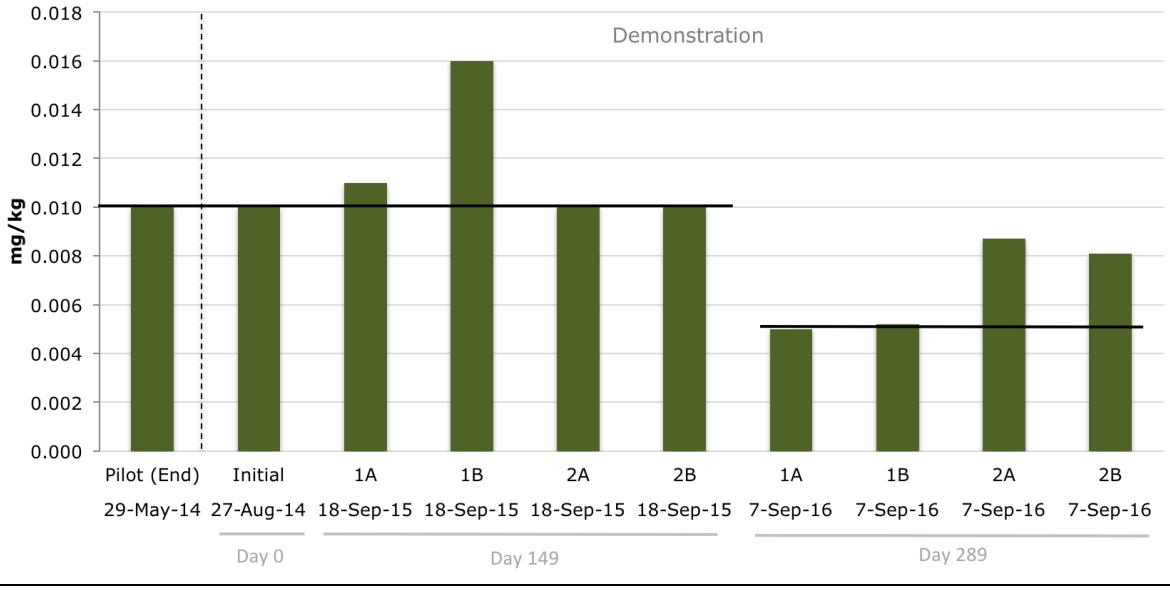
Figure 26 – Continued.
Pictures are of cell 1B.



Figure 27 – Aphids observed in cell 1A and 2A.
Left picture is of cell 1A. Right picture is of cell 2A. Note the increased yellowing in cell 2A due to a denser aphid population.



Figure 28 – Aphids observed on Carex (September 8, 2016).
The top picture is of cell 2A, the bottom picture is of cell 1B. Note cell 2A has more yellowing due to larger aphid population.



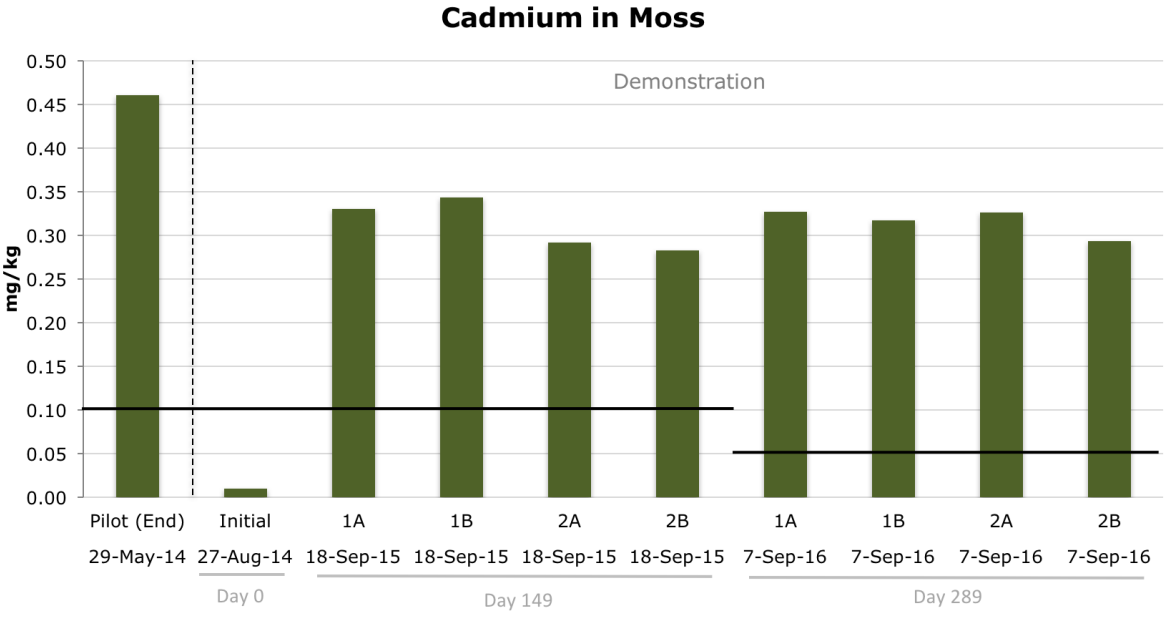
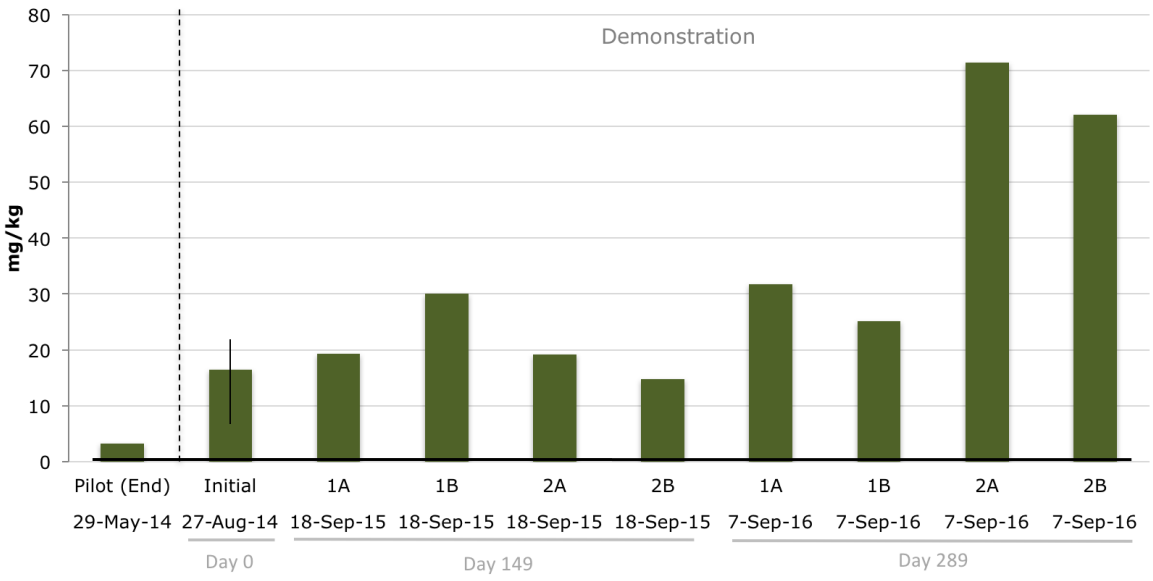


Figure 29 – Cadmium concentrations in plant tissue.
The 2014 and 2015 DL for Cadmium is 0.10 mg/kg. The initial data set is the average value of CWTS 1A-2B. The error bars indicate the minimum and maximum values in that data set.



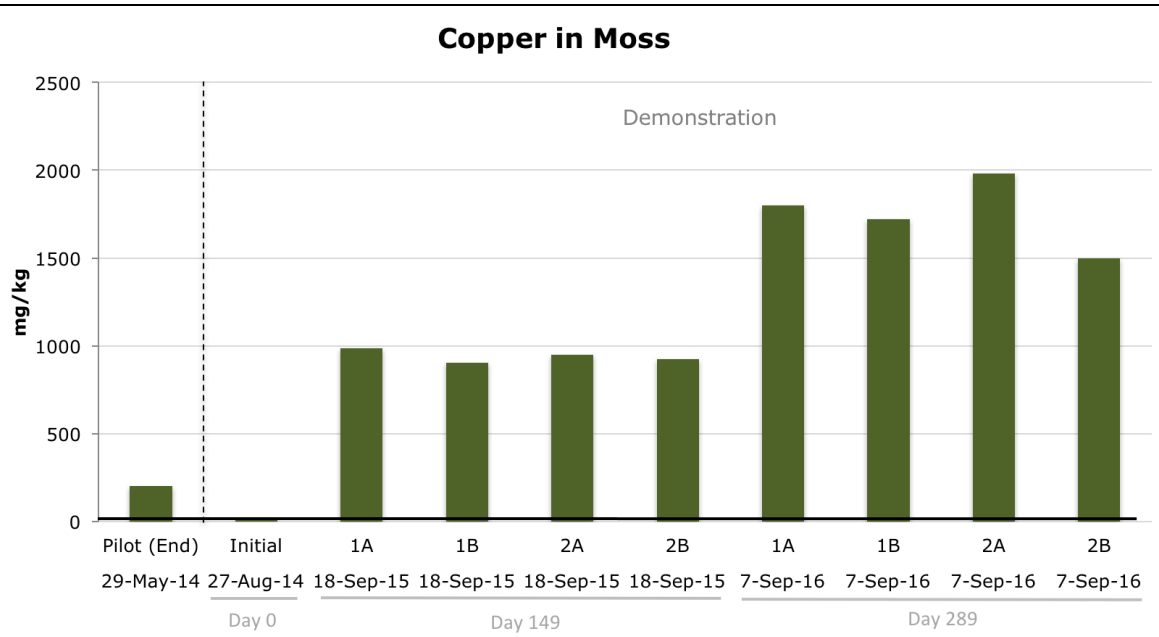
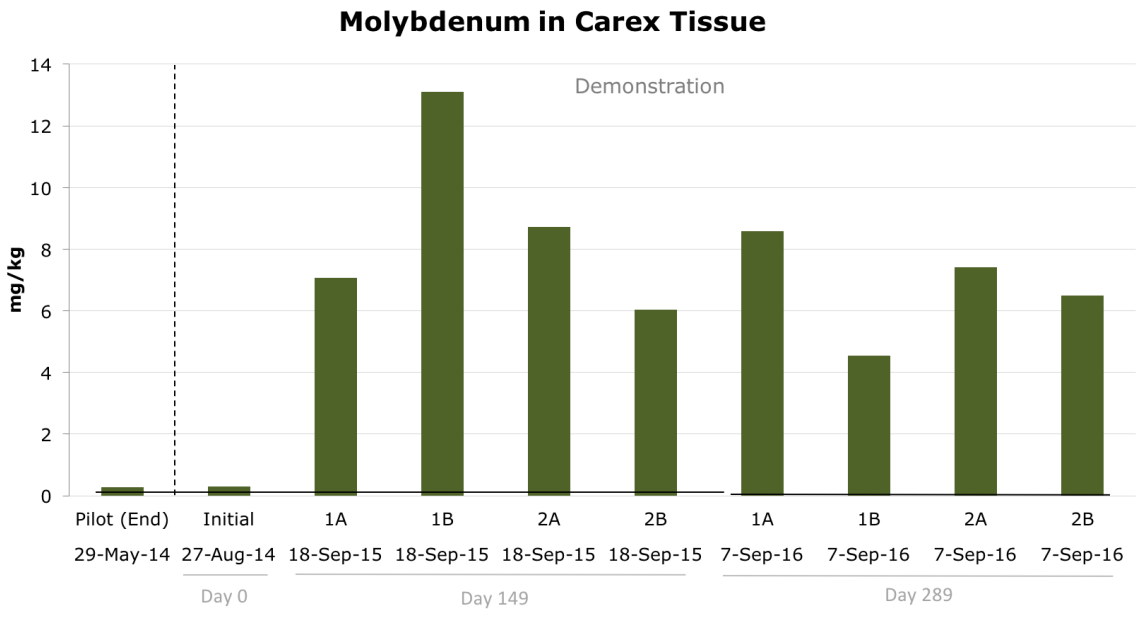


Figure 30 – Copper concentrations in plant tissue.
The 2014 DL for Copper is 0.5 mg/kg, the 2015 DL is 0.1 mg/kg, and the 2016 DL is 0.05 mg/kg. The initial data set is the average value of CWTS 1A-2B. The error bars indicate the minimum and maximum values in that data set.



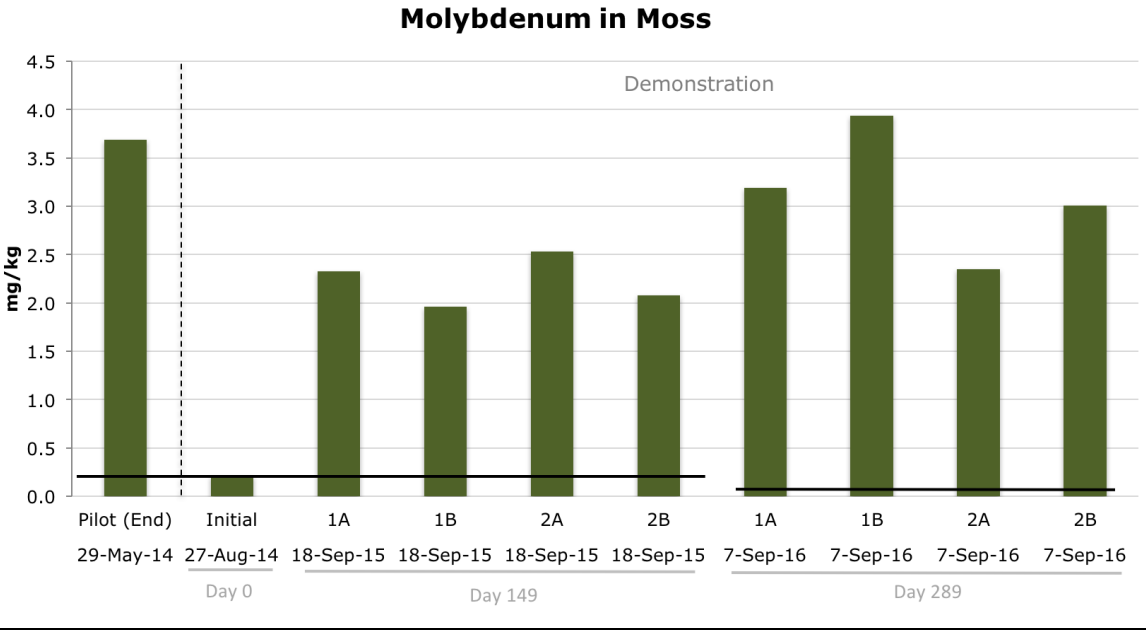
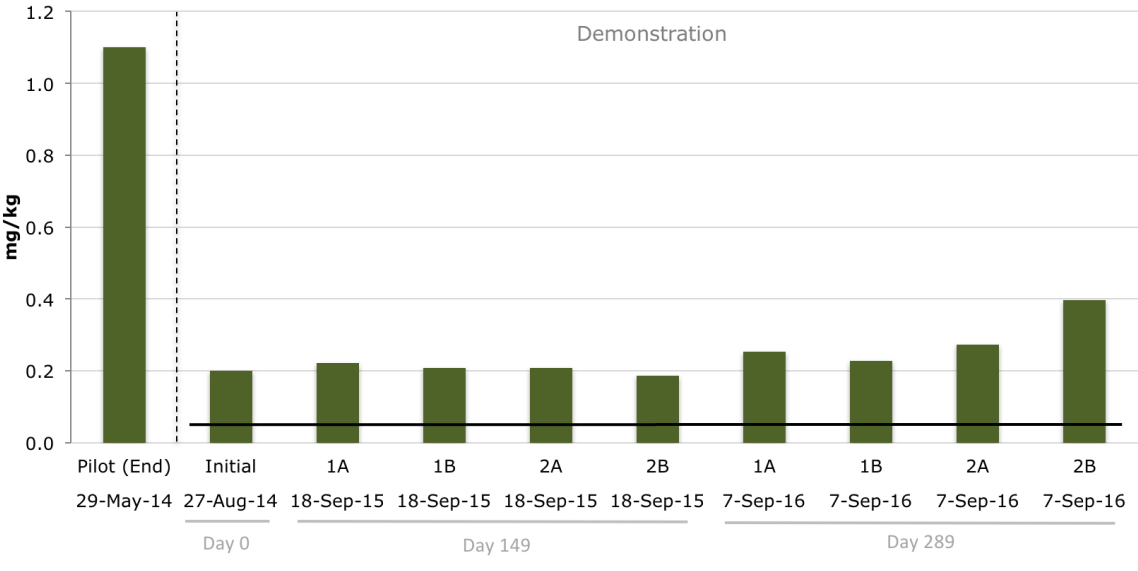


Figure 31 – Molybdenum concentrations in plant tissue.
The 2014 and 2015 DL for Molybdenum is 0.05 mg/kg, the 2016 DL is 0.020 mg/kg. The initial data set is the average value of CWTS 1A-2B. The error bars indicate the minimum and maximum values in that data set.



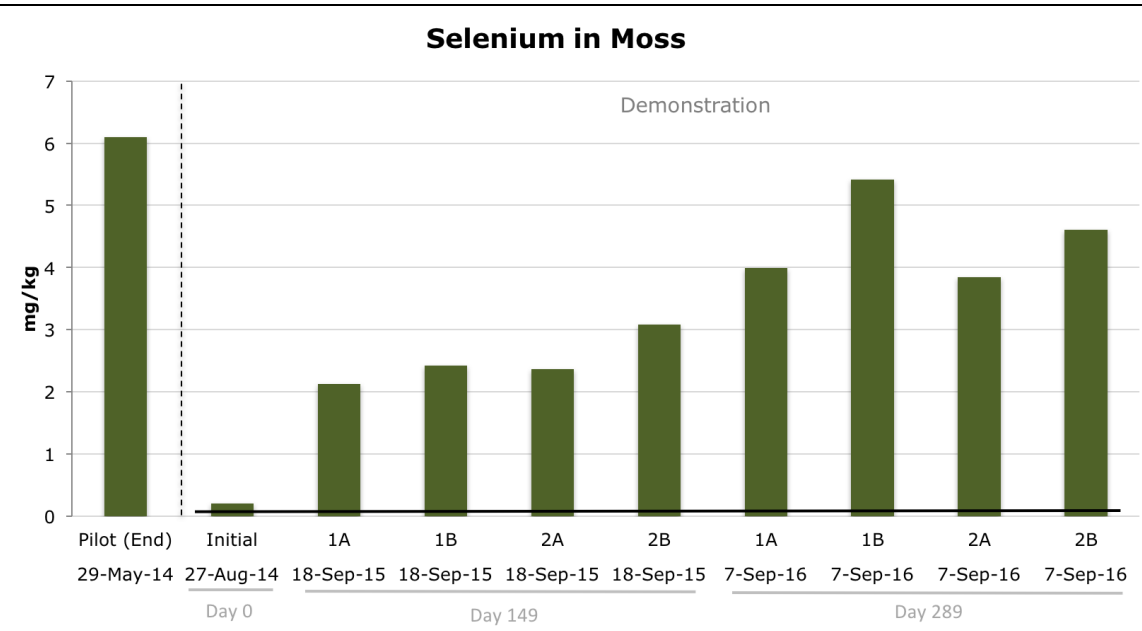
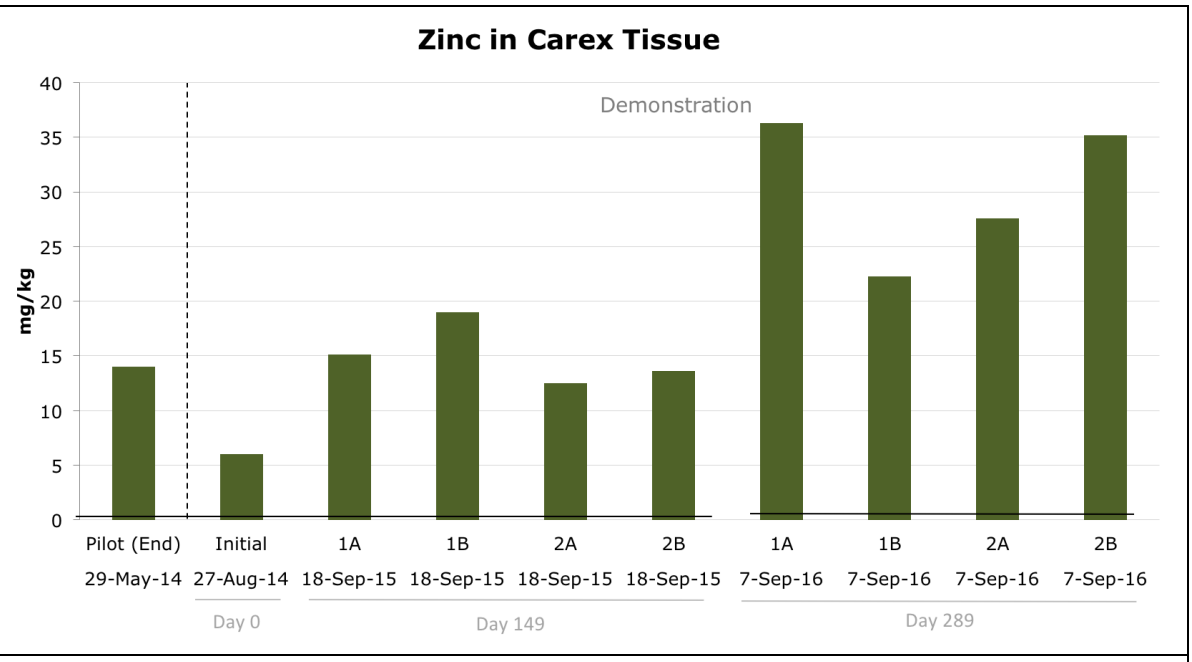


Figure 32 – Selenium concentrations in plant tissue.
The method detection limit for Selenium is 0.2 mg/kg. Initial data set is the average value of replacates used for planting of all CWTS cells. Error bars indicate the minimum and maximum values in that data set.



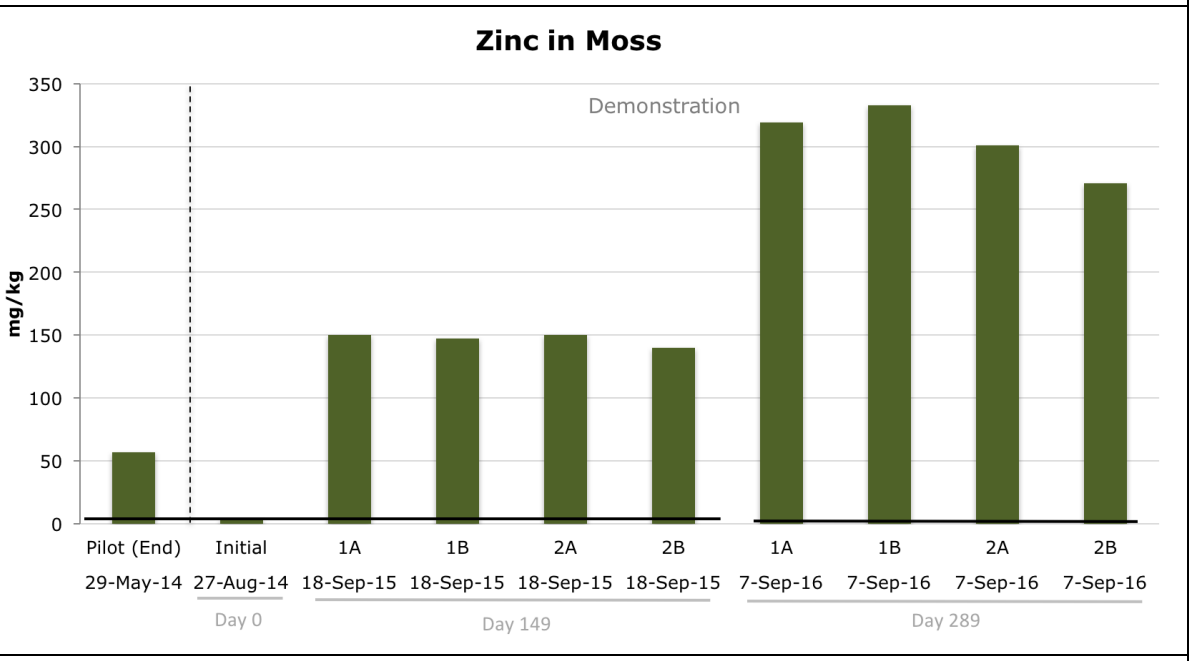


Figure 33 – Zinc concentrations in plant tissue.
The 2014 DL for Zinc is 2.0 mg/kg, the 2015 DL is 0.002 mg/kg, and the 2016 DL is 0.5 mg/kg. The initial data set is the average value of CWTS 1A-2B. The error bars indicate the minimum and maximum values in that data set.

10.3. Beneficial microbes

Microbes are the driving force of many treatment pathways that are targeted in CWTSs. The beneficial microbes in these systems catalyze biogeochemical cycles for remediation of specific constituents of concern. Careful design of CWTSs can create the environmental conditions needed to enhance the abundance and metabolic activity of these beneficial

microbes. Accordingly, complimentary methods of genetic and growth-based testing were used to characterize the microbial populations associated with a range of microbial habitats in the demonstration-scale CWTS (e.g., soils, sediment, biofilms, aquatic mosses, and plant roots).

In the context of the Minto Mine CWTS, beneficial microbes include those that are involved in the reduction of selenium (selenate and selenite), nitrate, and sulphur compounds (which in turn can treat copper, cadmium, molybdenum, and zinc). Information on each of these mechanisms and the associated microbial populations in the demonstration-scale system is outlined in the following sections.

10.3.1. Sulphide-producing bacteria

Treatment is achieved by targeting the lithic biogeochemical sequestration of divalent metals through sulfide (i.e., S^{2-}, HS^-) precipitation as mineralized species (e.g., chalcocite [CuS], covellite [Cu_2S]). These sulfide bound species are relatively insoluble (CuS; $K_{sp}=10^{-16}$; Stumm and Morgan 1996), and are transferred from the water column into the CWTS soil as non-bioavailable fractions (Murray-Gulde et al., 2003; Huddleston et al., 2008). Moreover, similar reactions occur with cadmium and zinc, rending them non-bioavailable. As such, sulphide production is a key biogeochemical treatment mechanism for water treatment at Minto Mine. Sulphides can be created by beneficial microorganisms through the reduction of sulphur-containing compounds, such as sulphate, sulphite, thiosulphate, and elemental sulphur.

Based on the information gathered in pilot-scale testing, the targeted soil redox for the demonstration-scale CWTS is between -100 and -250 mV. This is in agreement with literature that indicates anaerobic conditions with relatively low ORP (-250 to -100 mV) are necessary for promoting anaerobic metabolisms in bacteria which oxidize organic matter, producing electrons which reduce sulfate to hydrogen sulfide (H_2S) and other reduced sulfide species (i.e. bisulfide ion (HS^-), sulfide ion [S^{2-}]; Mitsch and Gosselink 2007). In these redox ranges, bacterial sulphide-production through reduction of sulphur compounds is expected, alongside increases in the proportion (percentage) and abundance of these microbes.

As expected, the proportion of sulphide-producing bacteria increased as the soil redox decreased in the demonstration-scale system through commissioning (Figure 34). The *C. aquatilis* roots were found to harbor high proportions of beneficial sulphide-producing bacteria (SPB), as well as a diversity of SPB (Table 12) which were different types than those found in the soil (Figure 34). This is good as it increases the diversity of bacteria in the CWTS that can carry out the beneficial reaction under a wider range of conditions than a small diversity of organisms could. Mosses, algae, and biofilms were typically associated with low proportions of sulphide-producing bacteria (data not shown), as would be expected in these photosynthetic and oxidizing environments.

Table 12 – Average number of different types of sulphide-producing bacteria in different sample types.

Sample Type	Before organics (all cells)	A cells post organic	B cells post organic
algae/biofilm	2	NT	NT
detritus	NT	6	8
moss	5	4	5
root	9	8	15
soil	7	11	15

NT – not tested. Post organic addition are samples collected in September 2016. The number of different types is based on counting the number of operational taxonomic units (clustered at 97% identity) that are classified as known sulphide-producing bacteria.

In addition to assessing the proportion of the community that are SPB, the inferred abundance of SPB was analyzed for all sample types (Figure 35). Soil and root samples were found to have the highest inferred abundances of sulphide-producing bacteria per gram of sample, while mosses typically had lower abundances. The black bottom part of the moss had ~100x more sulphide-producing bacteria per gram than the top part of the moss collected from the same location (Figure 24). This is desired as the mature mosses begin to show characteristics of the desired coupled transfer and transformation processes (with new biomass, and decomposition of older bottom; Section 10.1.2).

The overall abundance of sulphide-producing bacteria in the B cells have also increased in the soil and *C. aquatilis* root samples of the CWTSs over time, particularly in series 1 after the addition of organics (Figure 35). There was also an increase in the average number of different types of sulphide-producing bacteria in samples after the addition of organics (Table 12, Figure 34). The addition of organics was done to provide carbon sources needed by sulphide-producing bacteria and to provide the desired sulphur- and selenium-reducing conditions. It is anticipated the addition of organics will continue to provide the necessary carbon source to stimulate desired reducing conditions and microbial populations in 2017, and until the vegetation has established sufficiently to provide this carbon source annually going forward.

The microbial analysis of various sample types in the CWTS has therefore confirmed that the commissioning period is proceeding as expected, with beneficial sulphide-producing bacterial populations establishing in the soils alongside reducing conditions, which will continue to be monitored for stability or further improvement in 2017.

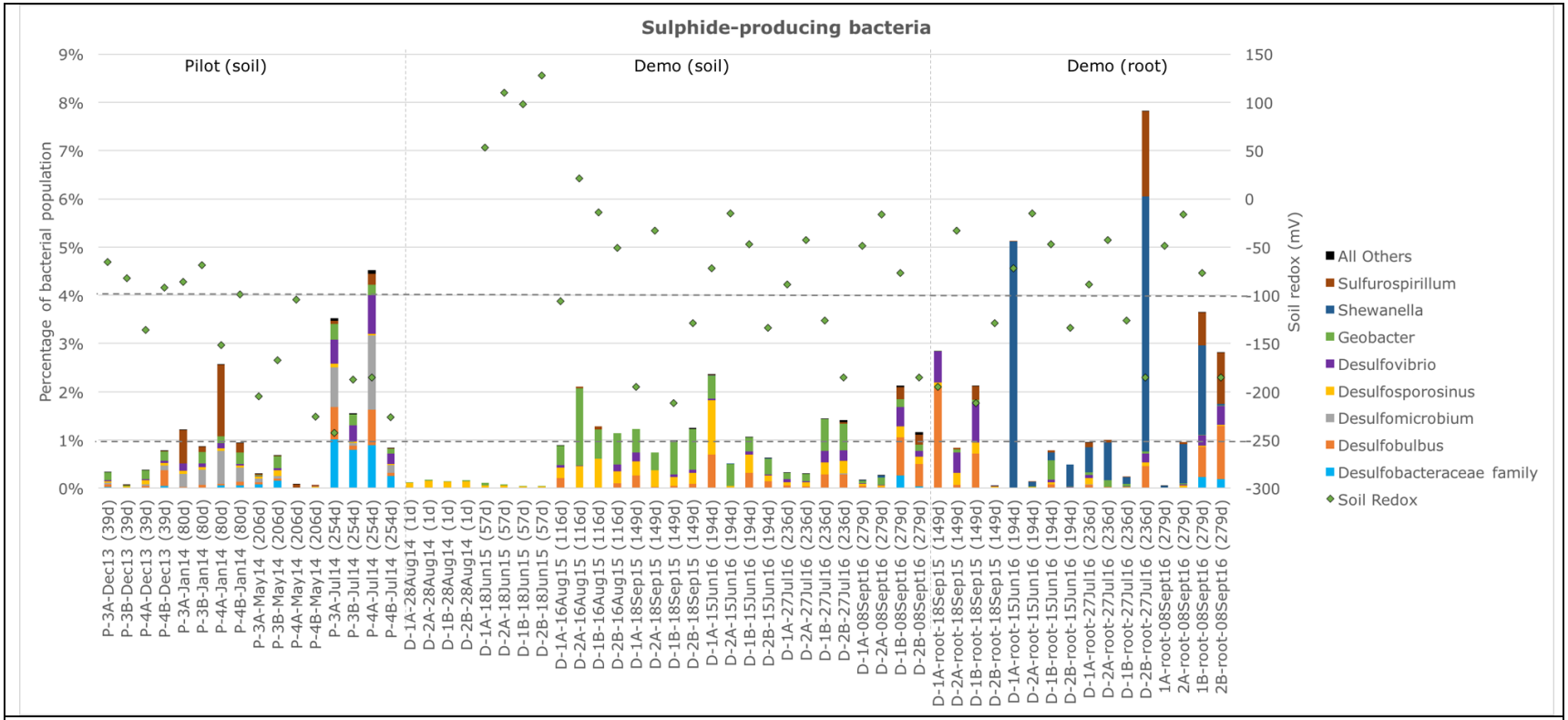


Figure 34 – Percentage and identity of sulphide-producing bacteria in soil of the pilot-scale and demonstration-scale CWTSs.

The vertical dashed lines separate the pilot-scale system data from the demonstration-scale system 1 and system 2 data. The horizontal dashed lines indicate the targeted soil redox range. Organism classifications and data analysis methods have been updated from Contango (March 2016).

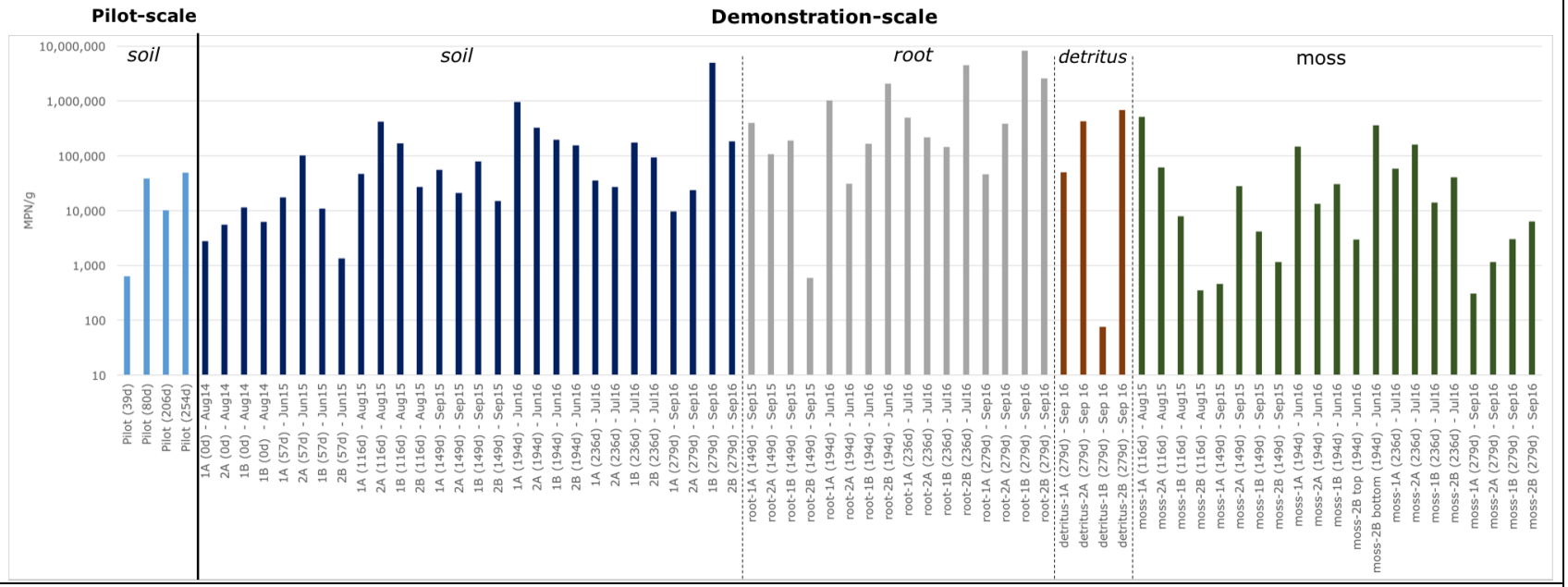


Figure 35 – Inferred abundance of sulphide-producing bacteria in soil of the pilot- and demonstration-scale CWTSs.

Inferred abundance is calculated based on a combination of genetic and growth-based data.
Shallow soil (0-5 cm); deep soil (10-20 cm); soil at 0d was that used for construction of wetland.

10.3.2. Selenium-reducing bacteria

The targeted selenium treatment pathways in the Minto CWTS include sorption to moss and soils, and subsequent microbial reduction of soluble (sorbed) selenate (Se(VI)) and selenite (Se(IV)) to insoluble elemental selenium (Se(0)). This reductive process can also be achieved directly in the water column, but is more effective associated with mosses and biofilms due to their sorptive properties that bring the selenium in contact with beneficial selenium-reducing bacteria. This is achieved within the range of soil redox conditions targeted for sulphate-reduction as suggested by pilot-scale testing and literature (see Section 5).

Selenite-reducing organisms are ubiquitous in nature and as expected, were detected in all sample types, including algae, biofilm, moss, soil, sediment, roots, and detritus. Although organisms that reduce selenate to elemental selenium (rather than intermediary selenium compounds) are generally less abundant in the environment, they were found associated with all sample types, indicating that the conditions conducive to their proliferation have been created within the CWTS. Moreover, the abundance of selenite- and selenate-reducing organisms increased over time in the demonstration-scale system during the commissioning period (Figure 36). Aquatic mosses were found to initially host the highest abundances of both selenate- and selenite- reducing organisms, affirming the importance of the inclusion of moss in the CWTS. Over time, as the vegetation has established, selenium-reducing bacteria have increased in abundance on the roots of *C. aquatilis,* and were also found associated with the added organic material once it began decomposing (Figure 36

These findings indicate the demonstration-scale CWTS has established beneficial selenium-reducing microbes in several areas of the wetland, which would be involved in selenium removal in the CWTS (either by reducing selenium that has been sorbed to moss or detritus, or by interacting directly with selenium in water that has been drawn into the root zone by plants). Their abundance is similar to what was found during pilot-scale testing in the soil, indicating they have established as expected. Selenium-reducing microorganisms will continue to be monitored in 2017, alongside performance testing.

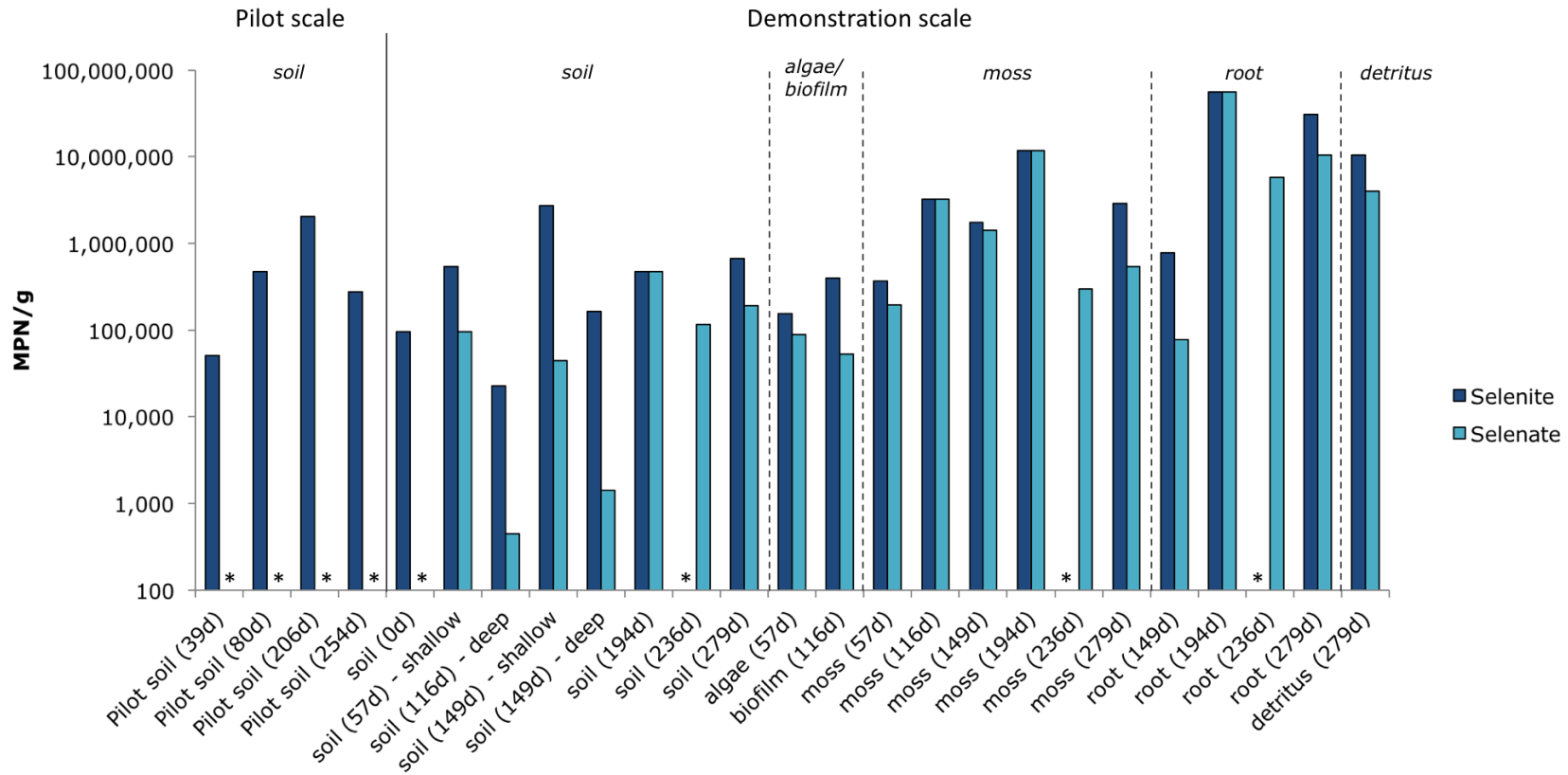


Figure 36 – Abundance of selenite- and selenate-reducing organisms in various CWTS sample types over time.
* Selenite or selenate not tested at this timepoint
Shallow soil (0-5 cm); deep soil (10-20 cm); soil at 0d was that used for construction of wetland. Organics material was added to the wetland after day 236.

10.1. Nitrate-reducing bacteria

Nitrate is sometimes a constituent of concern during operations and early closure owing to residuals from blasting activities. Even if not in exceedance of water quality guidelines in terms of receiving environment objectives, nitrate often requires attention to achieve treatment of other constituents. The presence of nitrate can interfere with the treatment of certain elements in water (such as selenium). Nitrate can be removed from water by different types of microbes, including nitrate reducing bacteria which can reduce nitrate (NO_3) to nitrite (NO_2), and also denitrifying organisms that are capable of fully reducing nitrate to nitric oxide (NO), nitrous oxide (N_2O), and dinitrogen gas (N_2 - which is the most abundant gas in air). MPN analysis was therefore used to quantify these organisms.

Nitrate-reducing and denitrifying organisms were found associated with all sample types in the demonstration-scale CWTS (Figure 37). Roots and detritus had a high abundance of both nitrate and denitrifying organisms, with soil being similar to what was found during pilot-scale testing. These results indicate nitrate reducers have established in the CWTS during the commissioning period as expected. Nitrate-reducing microorganisms will continue to be monitored in 2017, alongside performance testing.

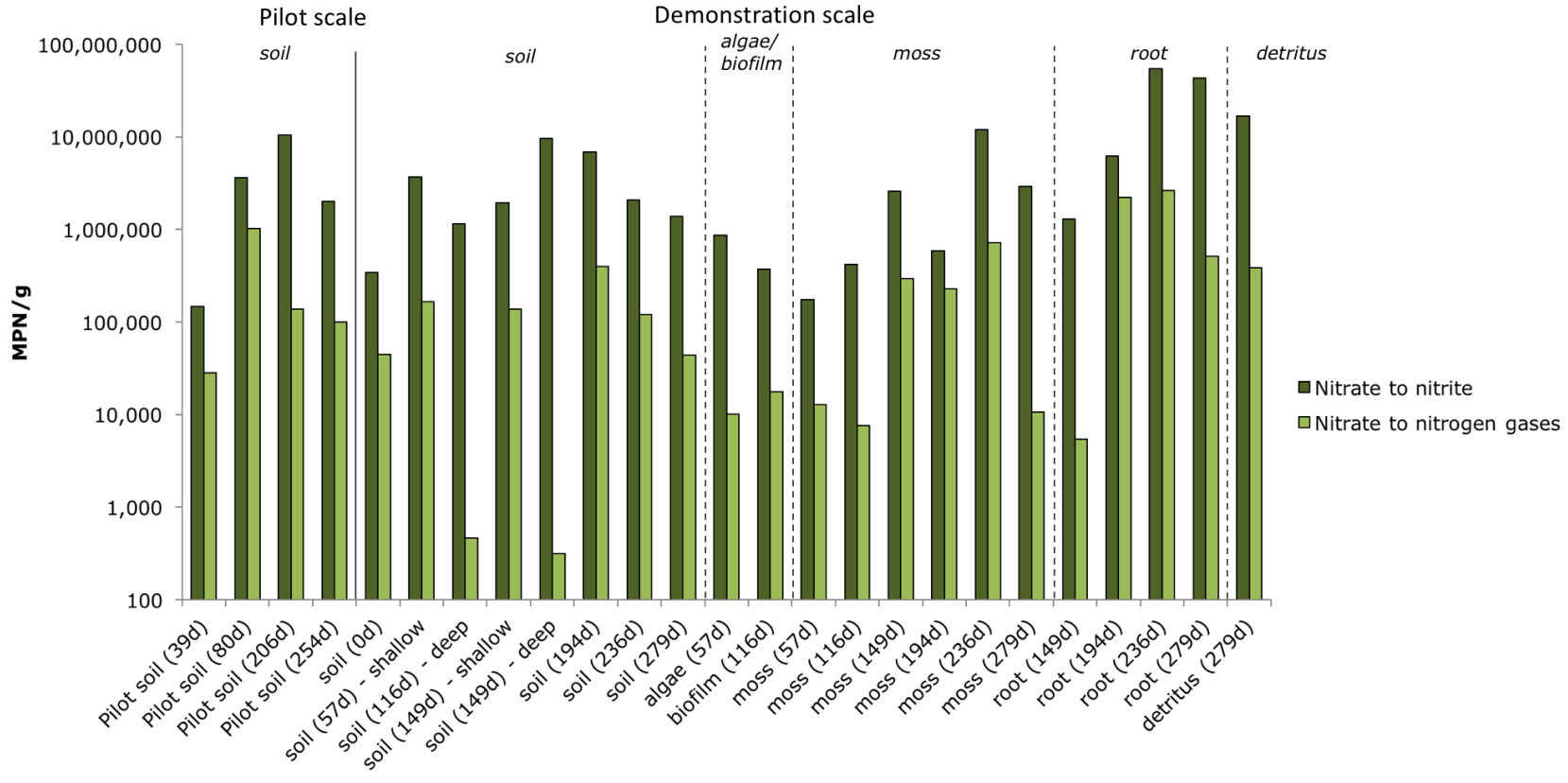


Figure 37 – Abundance of nitrate-reducing and denitrifying organisms in various CWTS sample types over time.
Nitrate-reducing = reduction of nitrate to nitrite; denitrifying = reduction of nitrate to nitrogen gases.
Shallow soil (0-5 cm); deep soil (10-20 cm); soil at 0d was that used for construction of wetland.

11.Summary of results

Table 13 summarizes results and findings from 2015 - 2016 commissioning of the demonstration-scale system at Minto.

Table 13 – Summary of Minto demonstration-scale 2015 - 2016 testing.

Objective	Purpose	Key Findings or Changes for Full-scale
Evaluate construction	Optimize construction and effectiveness of operation of full-scale systems	**Layout** -Outflow collection pond should have outflow at base (not top), with shutoff valve. -In 2016 sandbags were added at shores to prevent water short circuiting. Sandbags were also added to the outflow of series 1 to increase water depth.
		Soils -Use substrate with less total and leachable metals and metalloids (especially copper). -Higher sand content would improve hydrology, constructability (ability to level soils, ease of planting) and accessibility for sampling. -Organics should be mixed in bulk to soils prior to adding to cells.
Assess commissioning timelines	Allow for proper phasing of implementing full-scale systems for closure	**Water** -Commissioning-A period was complete at the end of July 2016 and the commissioning-B period (addition of organics to remediate the copper in substrates used) began July 29, 2016 and is ongoing. -Copper is being treated by the CWTS, and during the commissioning-B period the average influent concentration decreased by 37% compared to the average outflow concentration. -The wetland is achieving better treatment than suggested by inflow and outflow concentrations of the system, as soils are leaching aluminum and copper into the water. -Cadmium and zinc are also being treated in the CWTS, and during the commissioning-B period, the average outflow

Objective	Purpose	Key Findings or Changes for Full-scale
		concentrations were 64% and 69% lower than average influent concentrations respectively. - Molybdenum and selenium experienced notably greater treatment after during the commissioning-B period. The average outflow concentrations were 21% and 41% lower than average influent concentrations respectively. -The wetland is maturing as expected and is performing beyond anticipated based on the design. - Cadmium, copper, molybdenum, and selenium % removal and *k* rates are increasing over time. Zinc % removal is increasing over time. - HRT tracer study conducted in 2016 assessed hydrology and pore volume incorporation of the CWTS and determined HRT and removal rate coefficients for full-scale sizing. Rate coefficients should be re-evaluated after commissioning-B is completed. -Evapotranspiration study conducted to determine amount of evapotranspiration occurring in the CWTS in warmer months and to incorporate into load removal models. Due to rain the results were inconclusive and should be redone in 2017.
		Soils -Soil redox not consistently reaching targeted ranges by July 2016 therefore organics were added to stimulate desired reducing conditions, which marked the end of commissioning-A and the beginning of commissioning-B. -Cell 2B consistently reaching targeted ranges by the end of 2016. -In 2015 significant amounts of metals were leaching from soil substrate into water, putting additional treatment demands on system. In 2016 the amount of metals leaching from soil substrate into the water has decreased.
		Microbes -Sulphide-producing bacteria needed for copper and other metals removal have increased through the commissioning period as soil redox approached targeted ranges. Proportions are comparable to

Objective	Purpose	Key Findings or Changes for Full-scale
		those in pilot system at similar points in commissioning. After the addition of organics which marked the end of the commissioning-A period and the beginning of the commissioning-B period, sulphide-producing bacteria continued to increase, especially in the soil and roots in the B cells. -Abundance of selenium- and nitrate-reducing organisms are similar to those in pilot testing, indicating maturation as expected. -Selenium treatment performance increased through the commissioning period as mosses continued to grow, as they can sorb dissolved selenium and harbour highest abundance of selenate-reducing microorganisms to render the selenium insoluble. After the addition of organics which marked the end of the commissioning-A period, selenium-reducing bacteria were also associated with the added organic material once it started decomposing. Therefore the CWTS has established beneficial selenium reducing bacteria in several areas of the wetland (moss, detritus, roots, and soil).
Carex aquatilis transplantation effectiveness	Determine if plant propagation and/or replanting schedule will be needed for full-scale systems	->95% survival from transplanting. -Within first 2 months a further increase of >20%. -Full-scale system could be planted more densely to bring online faster, or less densely if time is less of an issue than sourcing plants (the plants are vigorous and will fill in the wetland in due time). - By the end of 2016 the plants had densely filled in the CWTS and transplanting was considered a success.
Moss colonization/distribution		-100% survival from transplanting. -Slower to spread, needs to be started more densely. -Staking helps maintain moss in 'upstream' parts of wetland, or could be transplanted multiple times through commissioning period. - Moss continued to mature in 2016 and does not appear to require additional transplanting.

12.Schedule and action items for 2017

Based on the performance results through 2015 and 2016, an action plan for CWTS optimization and testing for 2017 has been developed (Table 14). An updated multi-year schedule as per the proposed scope of work in the Minto Demonstration Scale Report Document 011_0315_01A (Contango, March 2015) is provided in Appendix C.

Table 14 – Minto 2017 CWTS demonstration-scale action items.

Who/When	Task	Contango Action Required	Additional Information
Minto Staff, prior to first site visit	Set flow rates at constant flow	Develop flow rate based on 2016 tracer study.	Do not change flow rate throughout year. Evaluate stability of wetland performance over one year, apply adjustments to models for full scale as necessary.
Contango site visit 7 (May/June 2017)	HRT Tracer study (salt)	Complete a second tracer study optimized from 2016 findings.	Coordinate an additional 1 m^3 tank at the CWTS site and a power source for the YSI.
	Aphid control (if needed)	Determine the appropriate method for aphid control.	
	General follow up	Complete seasonal sampling.	
Contango site visit 8 (July/August 2017)	Evapotranspiration study	Determine length of time and appropriate timeline to conduct the evapotranspiration study	Ensure the evapotranspiration study will not affect other planned activities at the site.
	General follow up	Complete seasonal sampling.	
Contango site visit 9 (September 2017)	General follow up	Complete seasonal sampling.	
Contango data monitoring throughout 2017 operations	Monitor metals leaching from soils	If metals continue to leach in 2017, flows could be stopped temporarily to promote this shift from oxidized to reduced forms. With the additional carbon sources of straw and wood chips now available for the bacteria to use to produce more sulphides for metals treatment	
	Monitor soil redox	Determine when it consistently reaches targeted range.	
December 2017	Report	2017 Update Report.	

13. 2017 Monitoring plan

A conceptual testing plan was developed for the demonstration-scale CWTS and has been refined and adapted based on performance and scientific findings. The demonstration-scale wetlands are expected to run until at least the end of 2018 in order to assess performance under a wider range of conditions. The conditions that are planned to be tested include both natural/environmental and selected influenced pressures, and can be imposed on the systems to mimic peak flow rates or droughts. In 2016 the systems completed the commissioning-A period and began the commissioning-B period, with plants becoming more established and abundant, microbial communities acclimating to the targeted conditions, and soil redox beginning to achieve targeted ranges. In 2017 the monitoring program will shift focus to testing of operations and performance. A multi-year plan is provided in Appendix C, and includes work performed to date as well as a schedule for 2017 and potential activities for 2018.

14. Closure

We trust the information herein satisfies your present requirements. Should you have any questions, please contact the persons listed below. We appreciate the opportunity to provide the services detailed in this report, and look forward to discussing any comments you may have.

Respectfully submitted,

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Appendix A

Table 1 - Summary of analytical sampling types, frequencies and locations.

Water	
Dissolved and total metals	Weekly, outflow of each cell and feed
Ammonia	
Nitrate	
Nitrite	
Flow rate	Weekly, feed
pH, DO, ORP, Conductivity (in situ)	Weekly, all cells and feed
Anion Sum	Monthly, outflow of each cell and feed
Cation Sum	
Hardness ($CaCO_3$)	
Ion Balance	
Total Dissolved Solids (TDS)	
Chloride (Cl)	
Sulphate (SO_4)	
Nitrogen (Total)	
Alkalinity	
pH	
Conductivity	
Chemical Oxygen Demand (COD)	
Total Organic Carbon (TOC)	
Total Kjeldahl Nitrogen (TKN)	
Total Suspended Solids	
Dissolved Organic Carbon (DOC)	
Bicarbonate (HCO_3)	
Carbonate (CO_3)	
Hydroxide (OH)	
Soil	
Relative soil redox (in situ)	Monthly, all probes (6 per cell)
Analyses completed on soil sample	
SAR, pH, EC, %sat, Ca, F, Mg, Na, K, Cl, SO4	Seasonally (3x per year) from a depth of 0-10 cm.
Available NPK and sulphur	
Alkalinity	
Bicarbonate (HCO_3)	
Carbonate (CO_3)	
Total Organic Carbon (TOC)	
Metals Analysis (Total)	
Sulfur by LECO	
Metals Leach	
AVS+SEM	

Sequential Leaching (5 Acid Test)	
Metals SPLP	Spring Sampling (1x per year).
Analyses completed on special leach	
Bromide (Br)	Seasonally (3x per year)
Chloride (Cl)	
Sulphate (SO_4)	
Fluoride (F)	
Alkalinity	
Bicarbonate (HCO_3)	
Carbonate (CO_3)	
Hydroxide (OH)	
Dissolved Organic Carbon (DOC)	
Ammonia	
Nitrate	
Nitrite	
Total Dissolved Solids	
Anion Sum	
Cation Sum	
Cation/EC ratio	
Ion Balance	
Plant tissue samples	
Metals Analysis	*Carex aquatilis* and aquatic moss, each cell, year end
Microbial samples	
Growth-based most-probable number analysis (nitrate reduction, selenite and selenate reduction, total heterotrophs)	Seasonally (3x per year)
Genetic sequencing analysis for bacterial community composition and distribution	Seasonally (3x per year)

Appendix B – Additional graphs

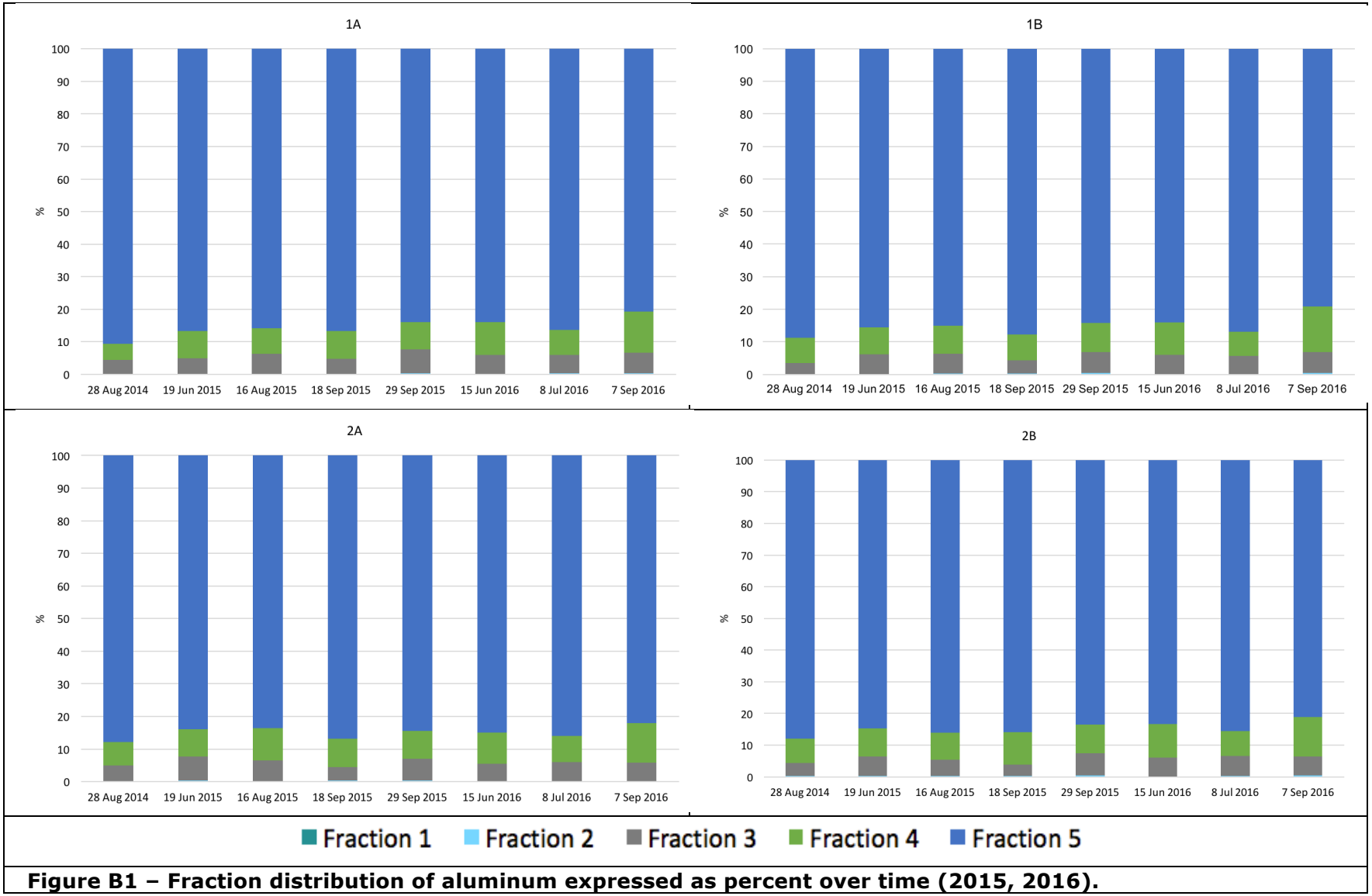


Figure B1 – Fraction distribution of aluminum expressed as percent over time (2015, 2016).

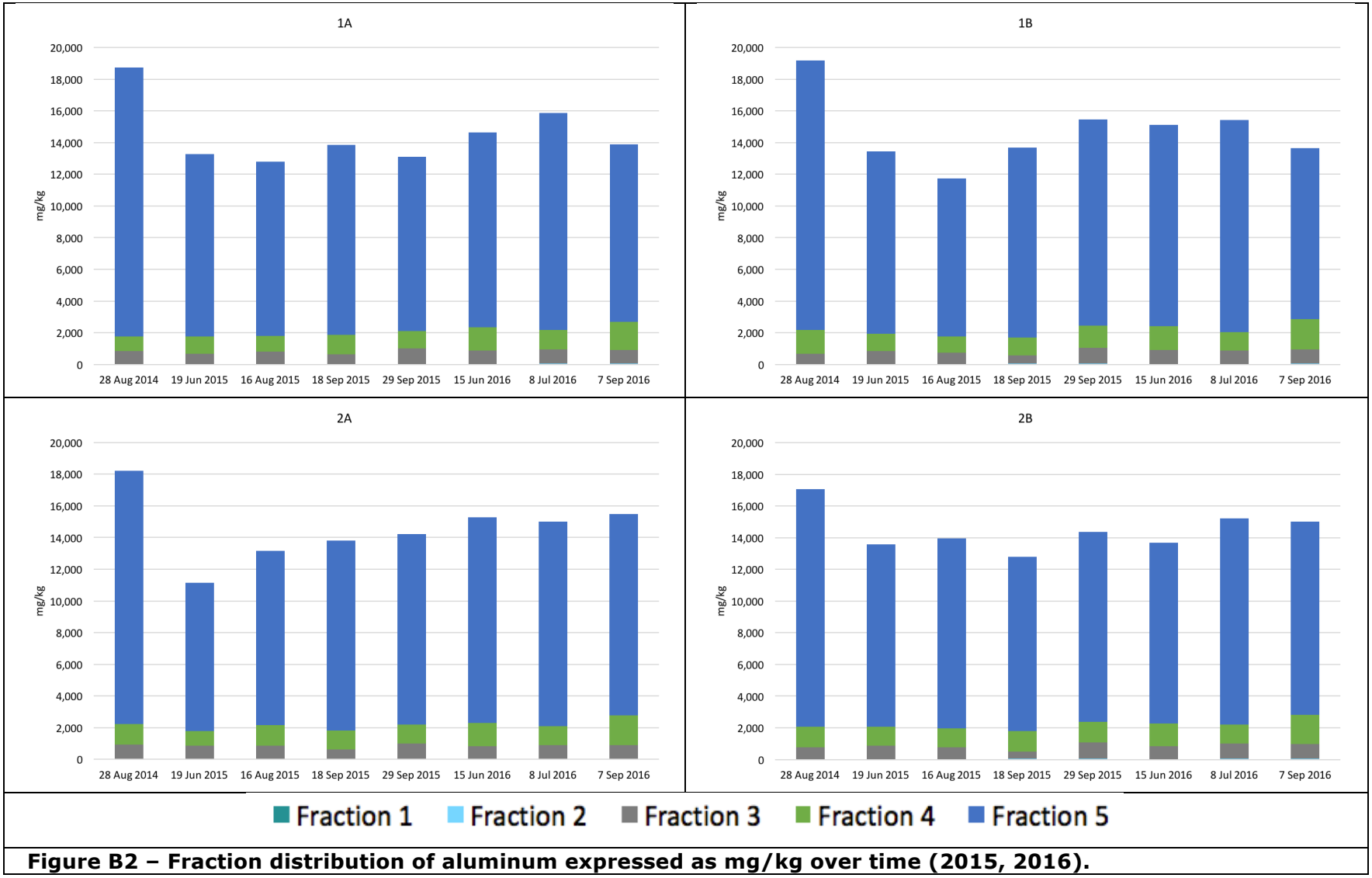


Figure B2 – Fraction distribution of aluminum expressed as mg/kg over time (2015, 2016).

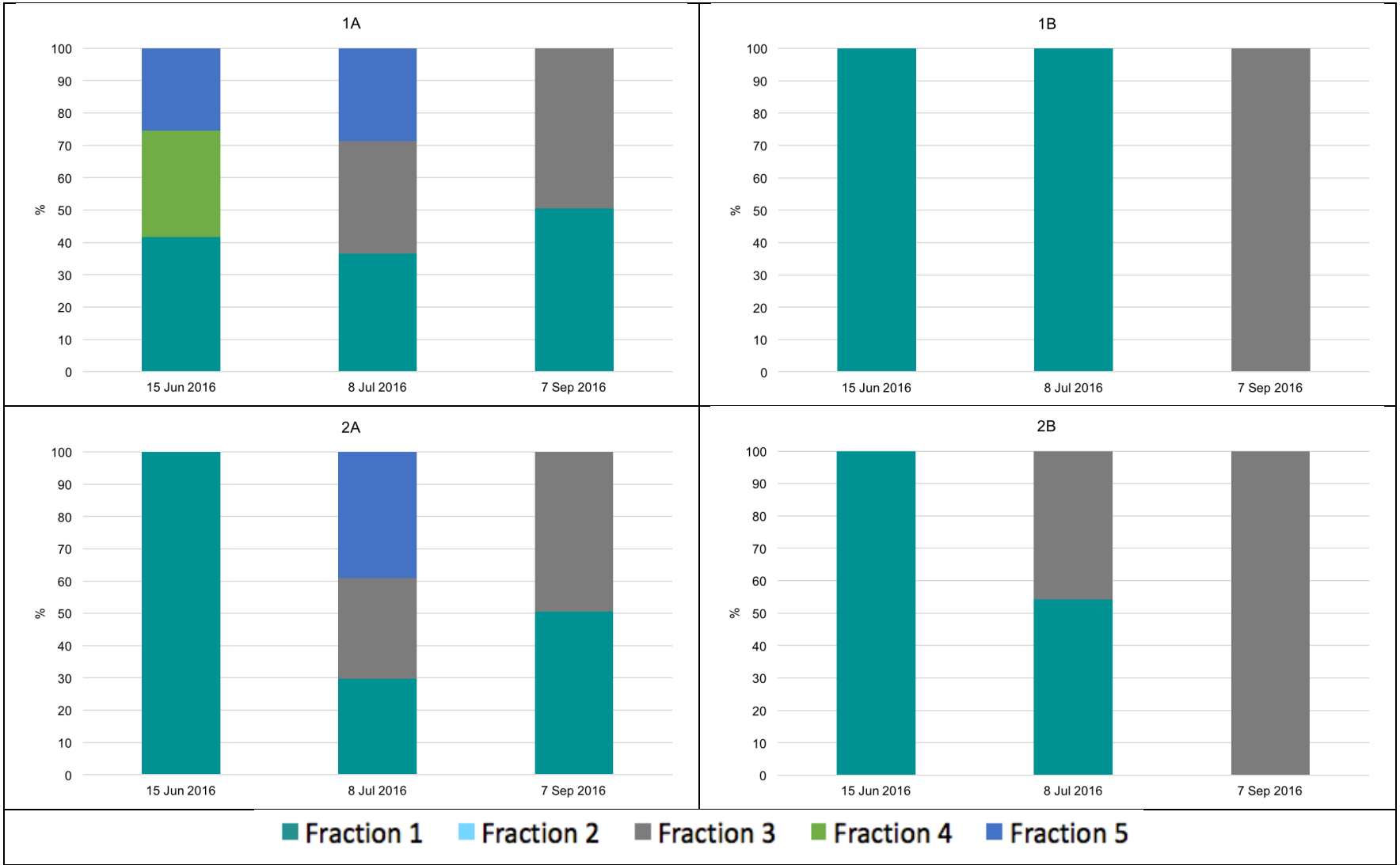


Figure B3 – Fraction distribution of cadmium expressed as percent over time (2016).
Data in 2014 and 2015 was below detection limits.

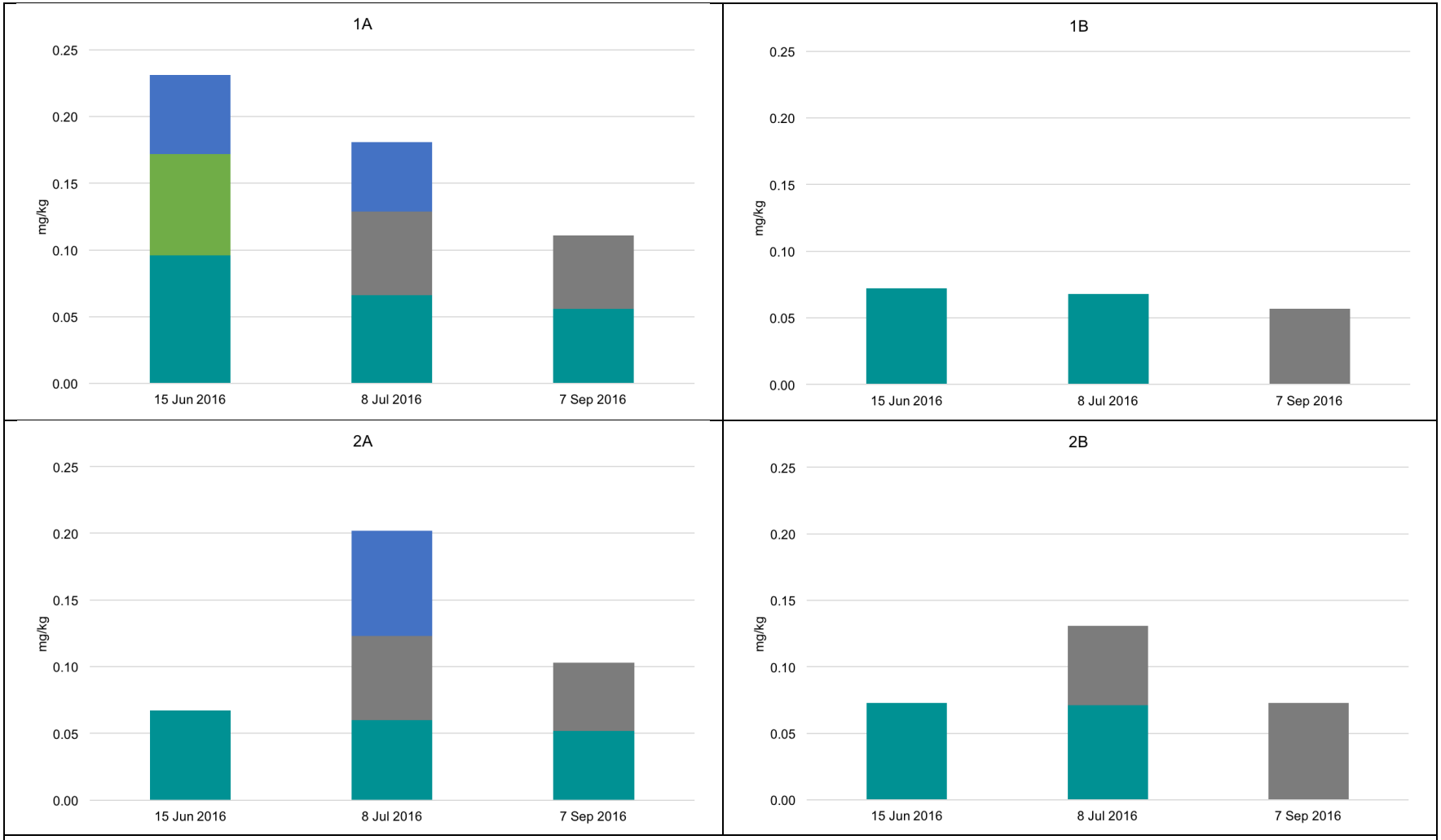


Figure B4 – Fraction distribution of cadmium expressed as mg/kg over time (2015, 2016).
Data in 2014 and 2015 was below detection limits.

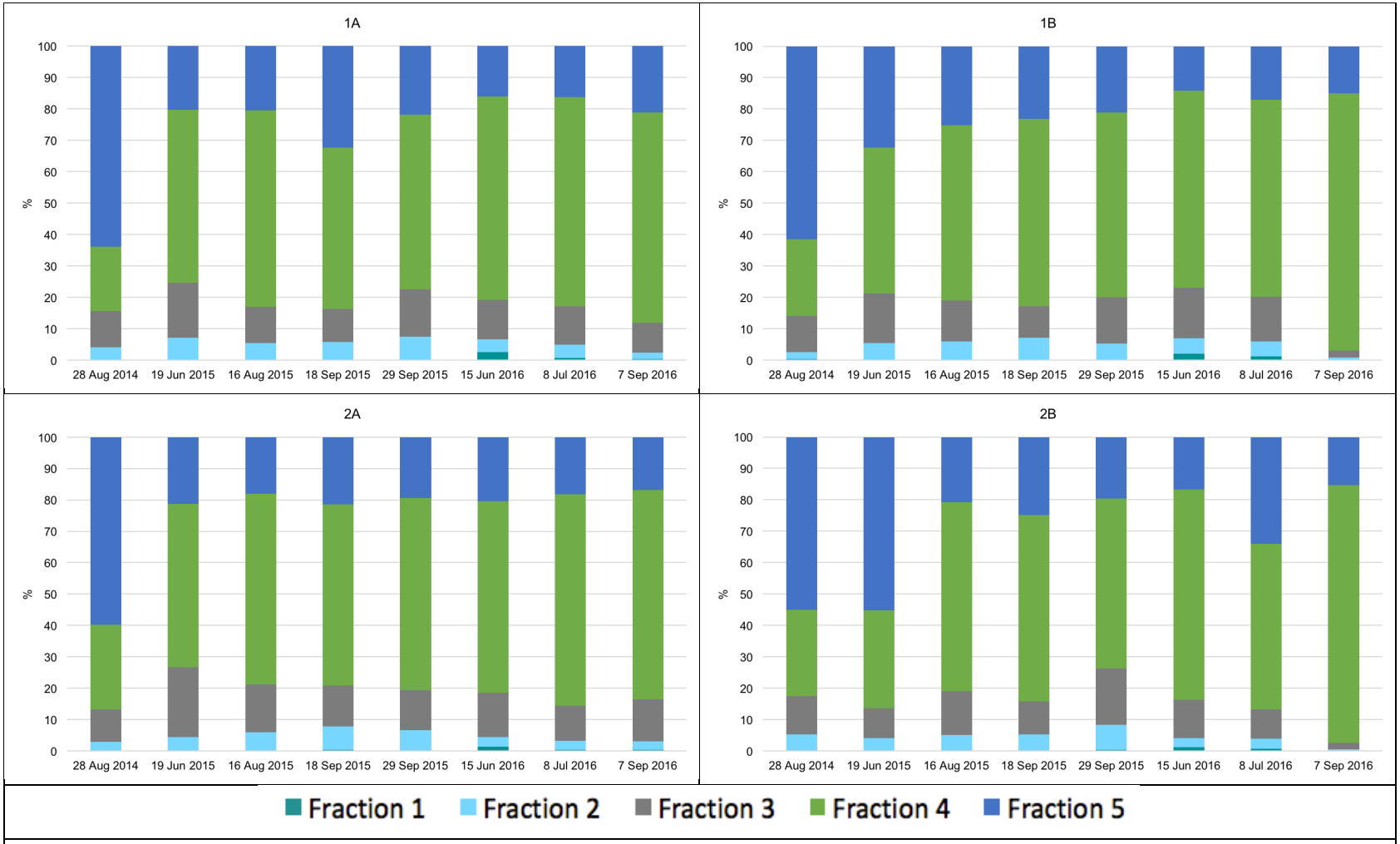


Figure B5 – Fraction distribution of copper expressed as percent over time (2015, 2016).

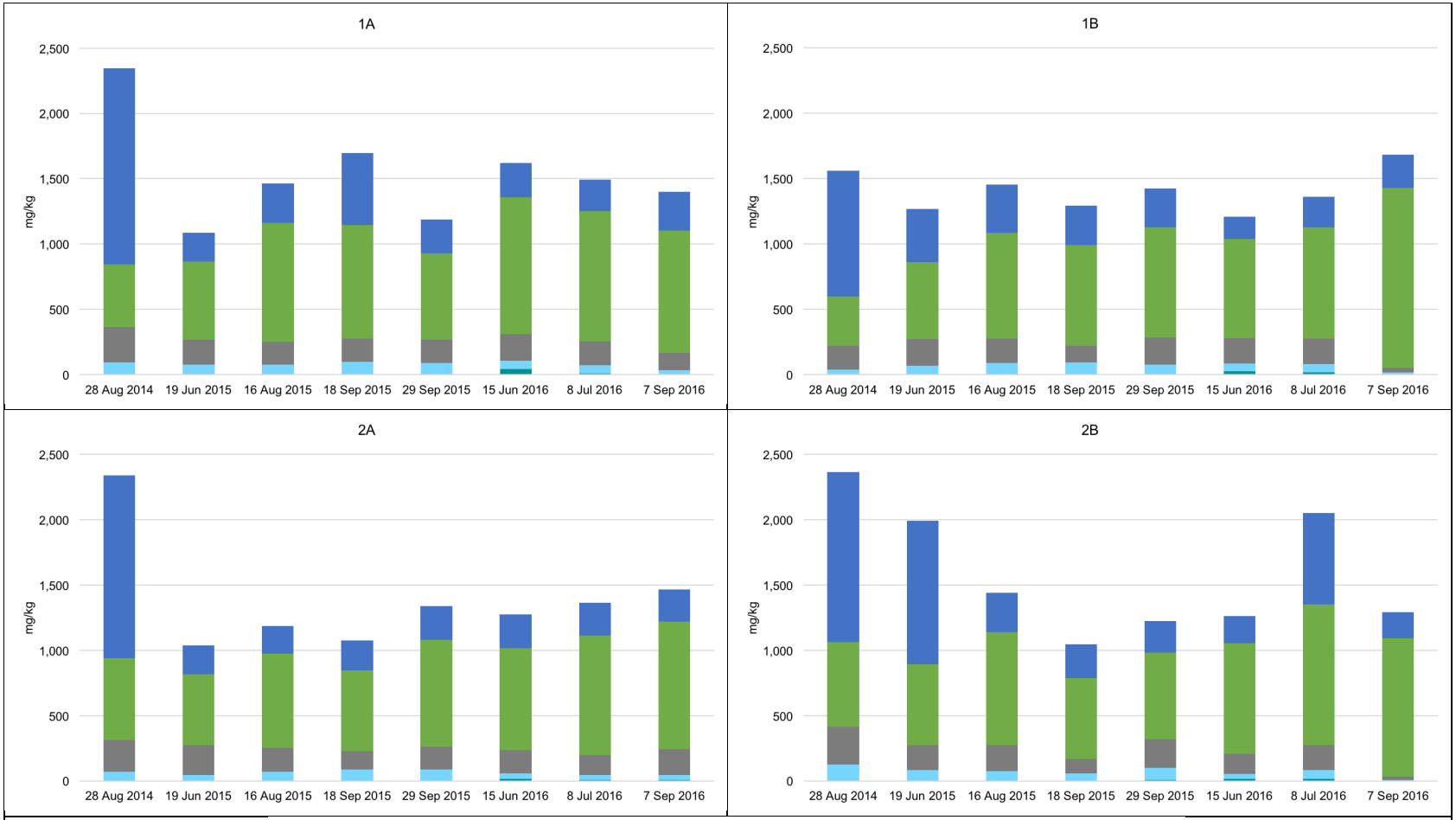


Figure B6 – Fraction distribution of copper expressed as mg/kg over time (2015, 2016).

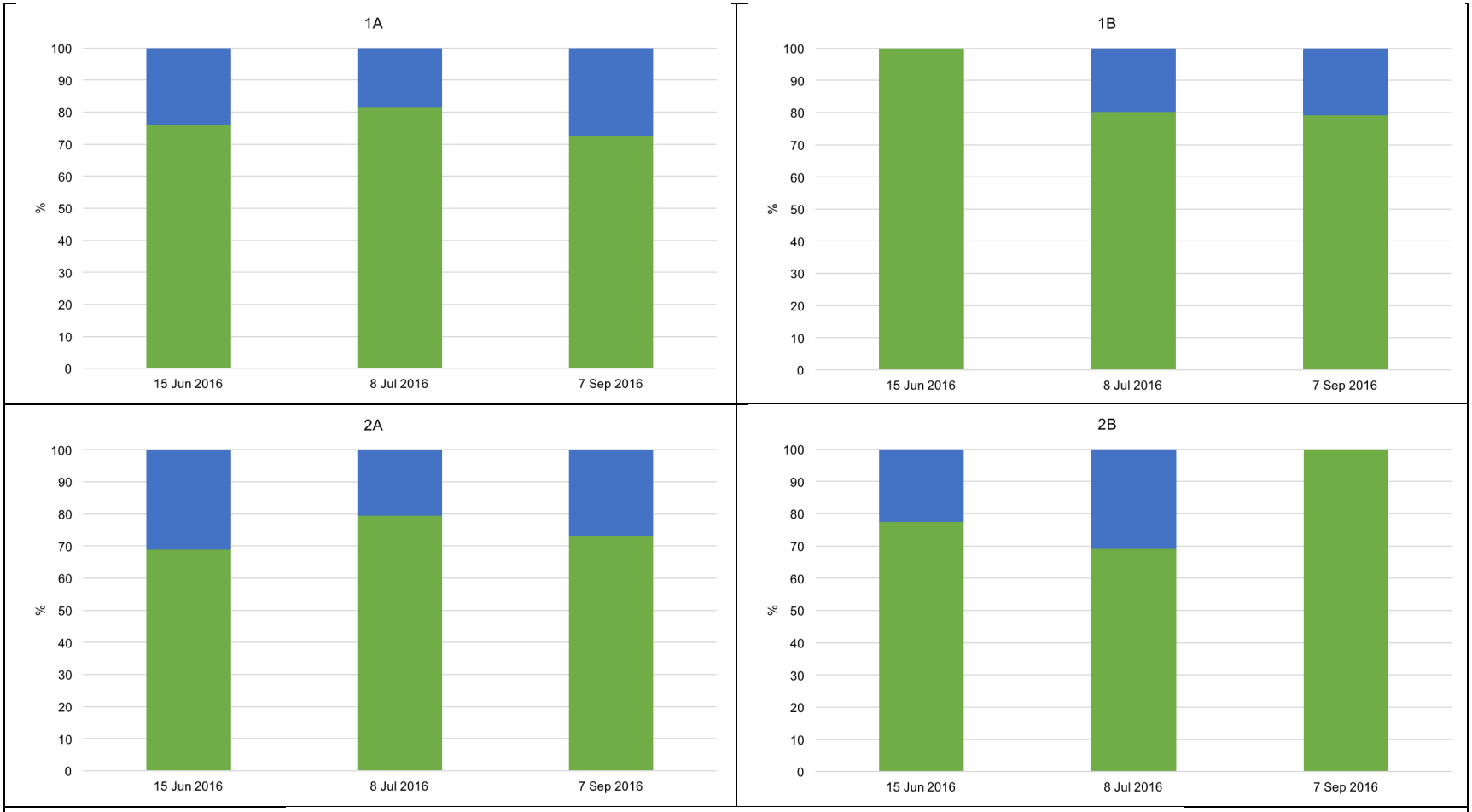


Figure B7 – Fraction distribution of selenium expressed as percent over time (2016).
Data in 2014 and 2015 was below detection limits.

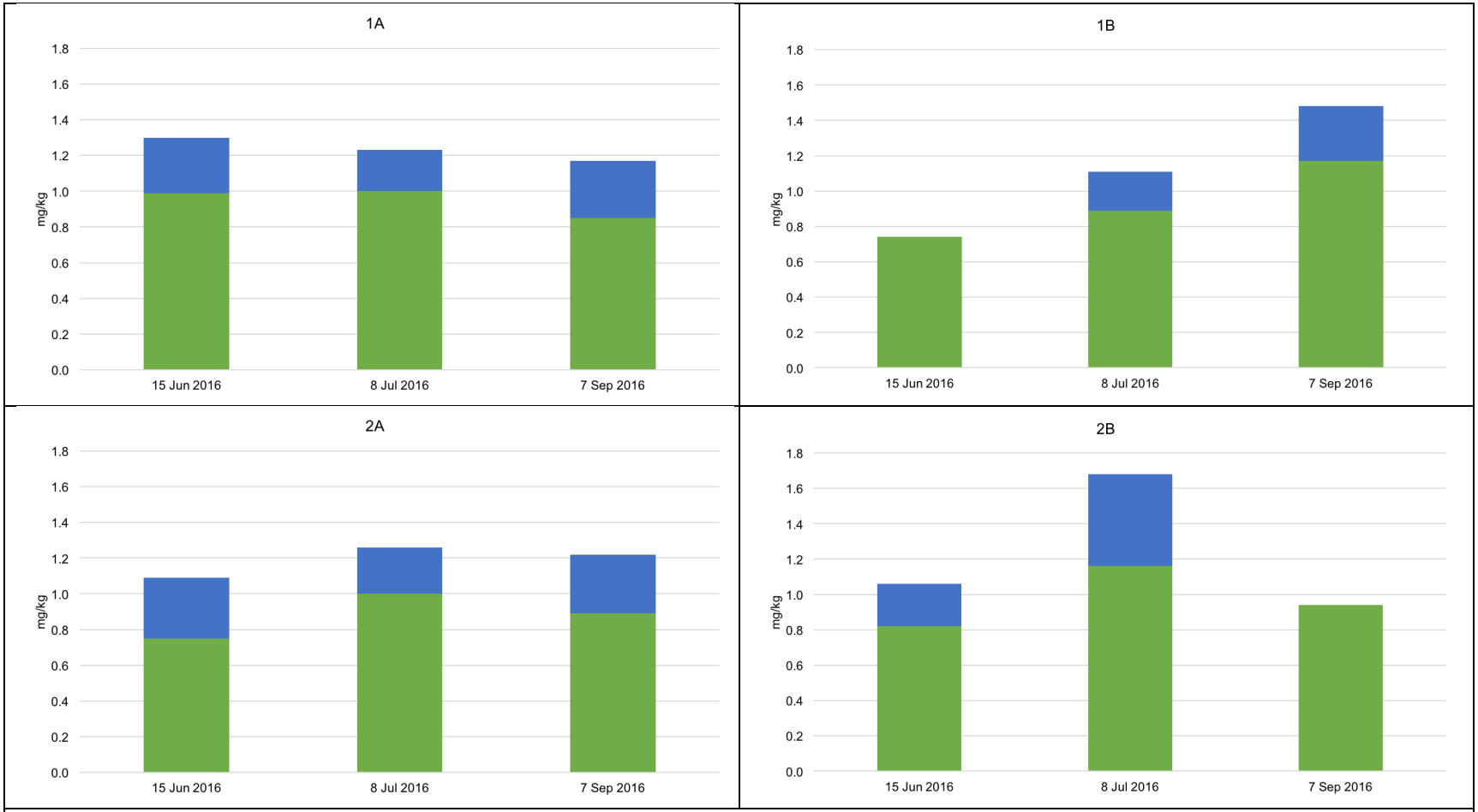


Figure B8 – Fraction distribution of selenium expressed as mg/kg over time (2015, 2016).
Data in 2014 and 2015 was below detection limits.

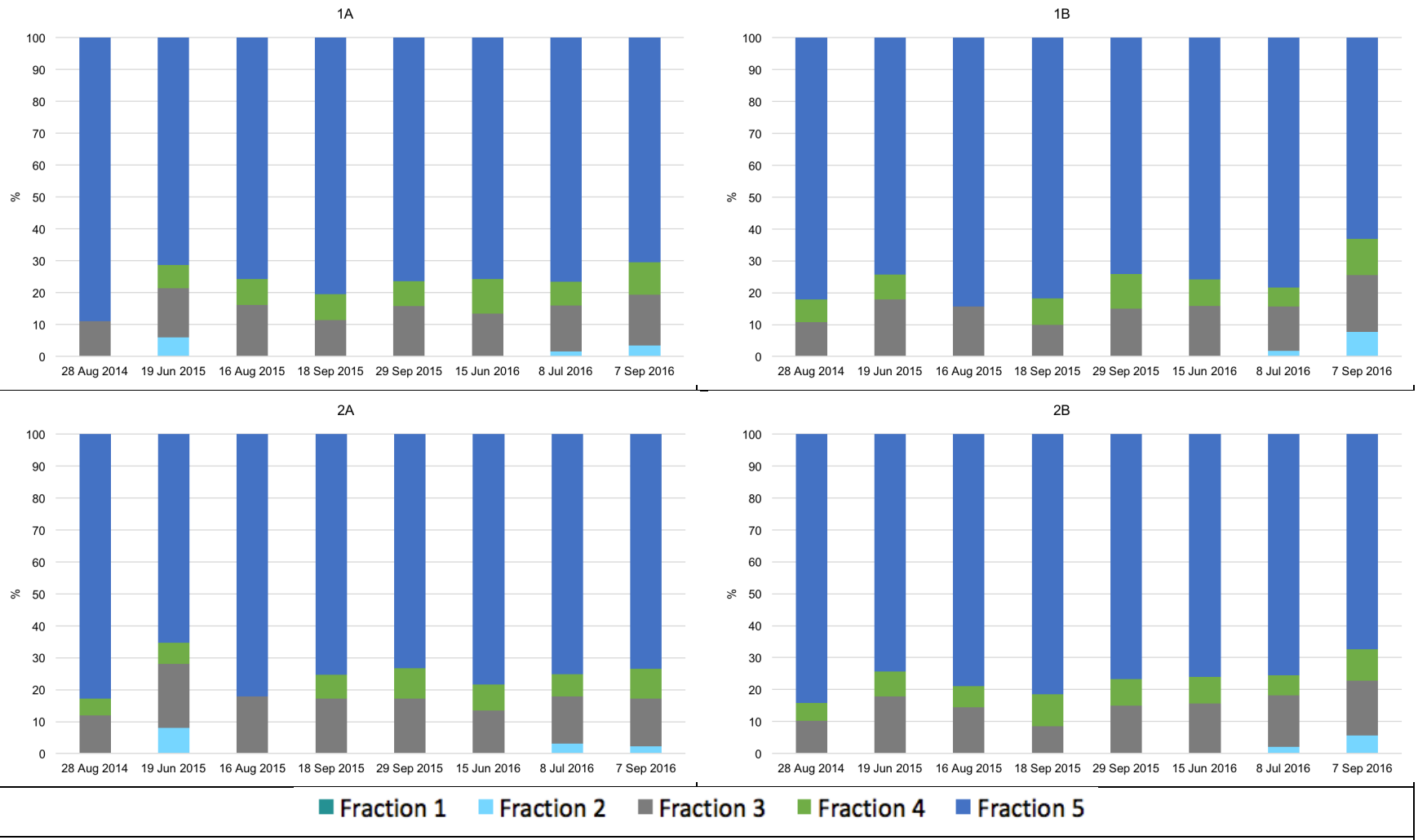


Figure B9 – Fraction distribution of zinc expressed as percent over time (2015, 2016).

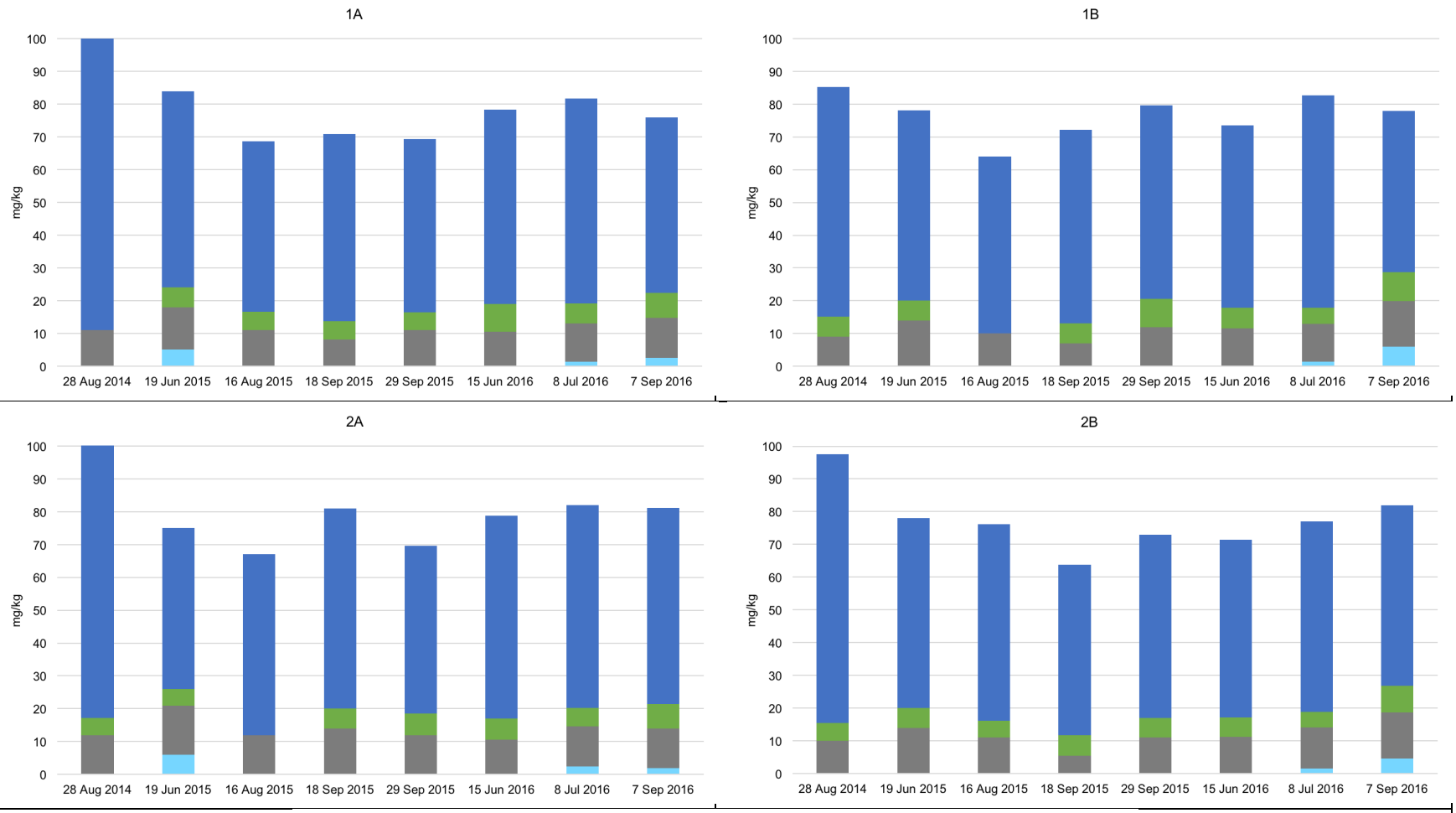
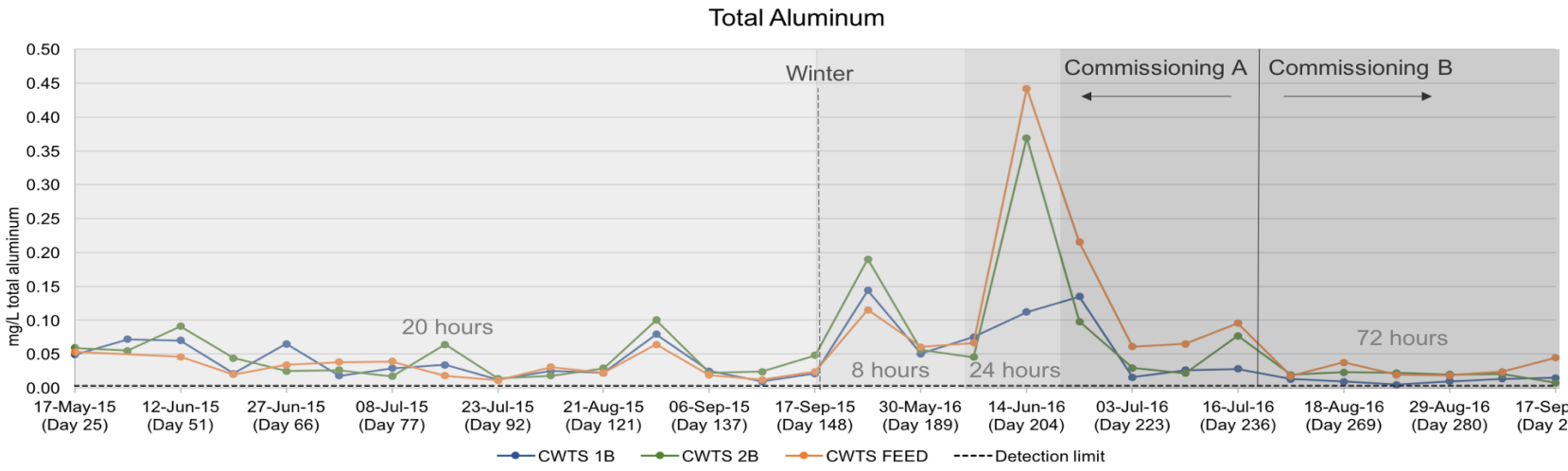


Figure B10 – Fraction distribution of zinc expressed as mg/kg over time (2015, 2016).



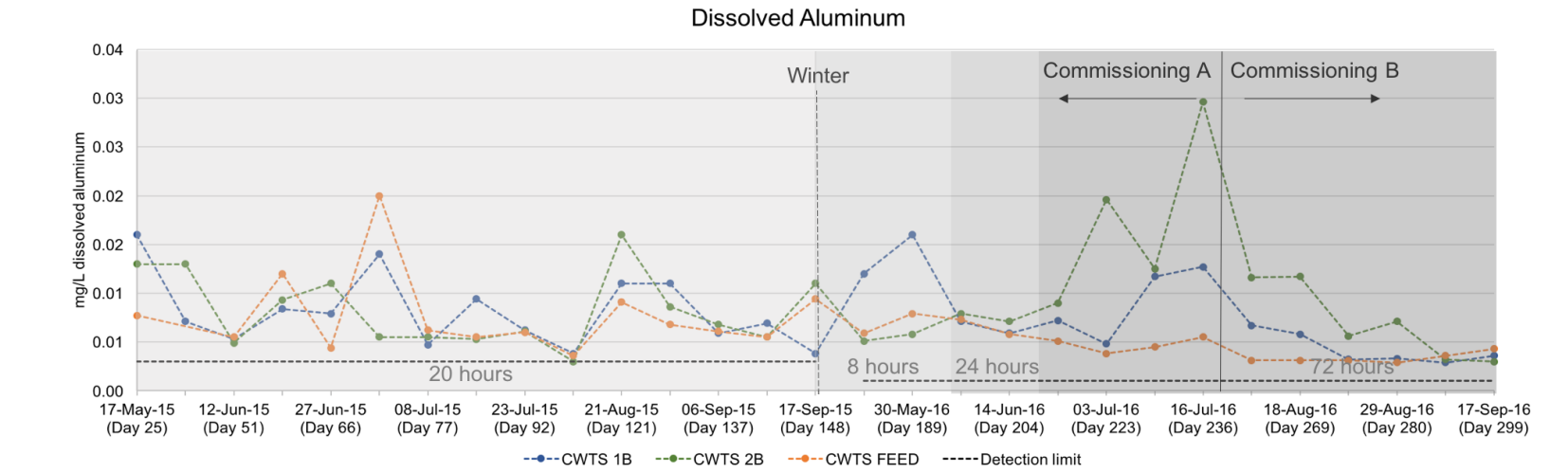
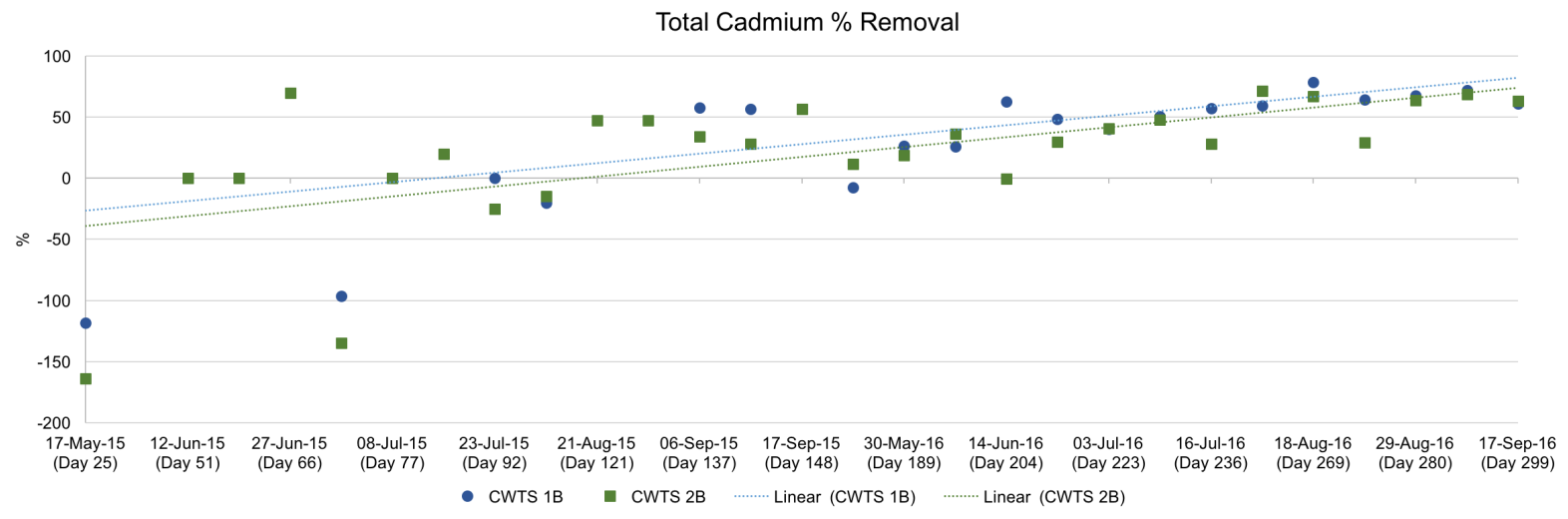


Figure B11 – Aluminum concentrations in the demonstration scale wetland from May 17, 2015 to Sept 7, 2016.
The HRT is indicated by grey shading and the duration in hours. The Maxxam (2015 results) measurement uncertainty (MU) for aluminum is +/- 0.0030 mg/L. The ALS (2016 results) MU is +/- 0.0010 mg/L. Timepoints where the dissolved concentration is higher than the total concentration are within the limit of uncertainty.



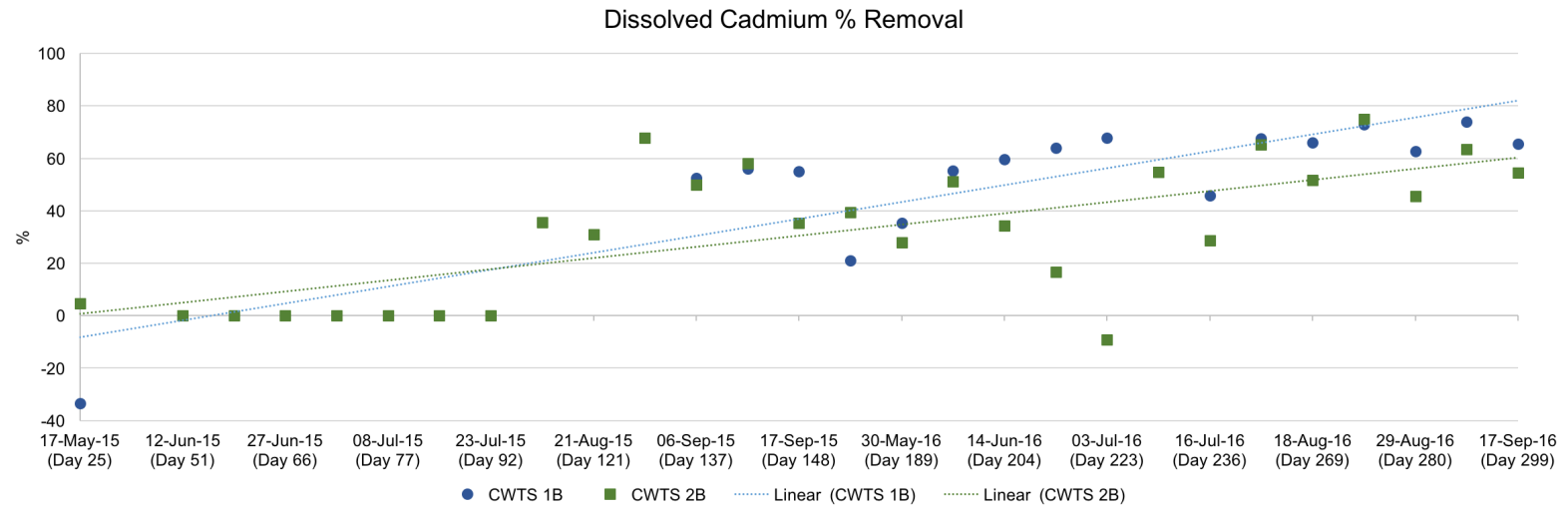
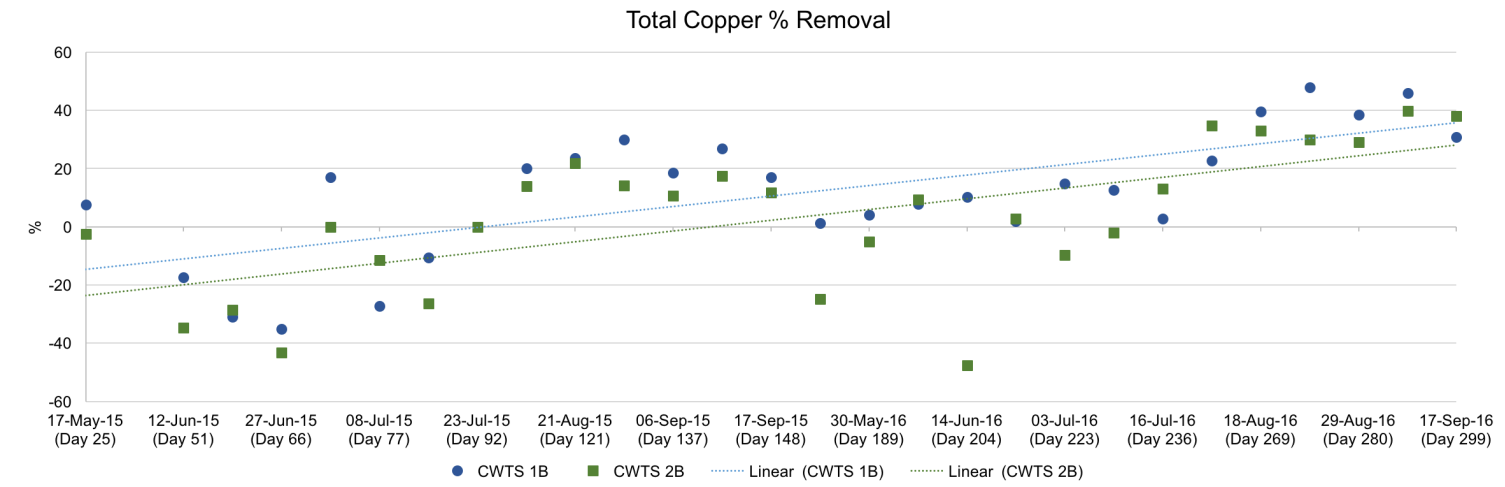


Figure B12 – Percent removal of cadmium in the demonstration scale wetland.
If the concentration was below the detection limit of the assay, the detection limit was used to calculate the percent removal. The Maxxam (2015 results) measurement uncertainty (MU) for cadmium is +/- 0.020 µg/L. The ALS (2016 results) MU is +/- 0.005 µg/L. The percent removal calculations for cadmium are therefore estimations within this range of error.



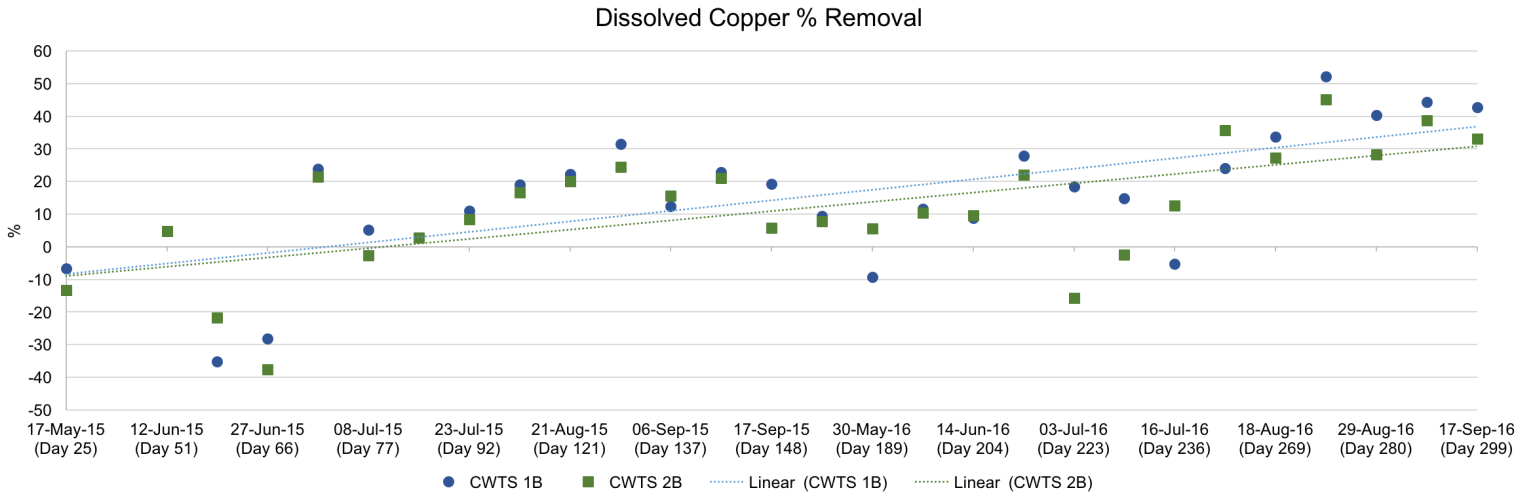
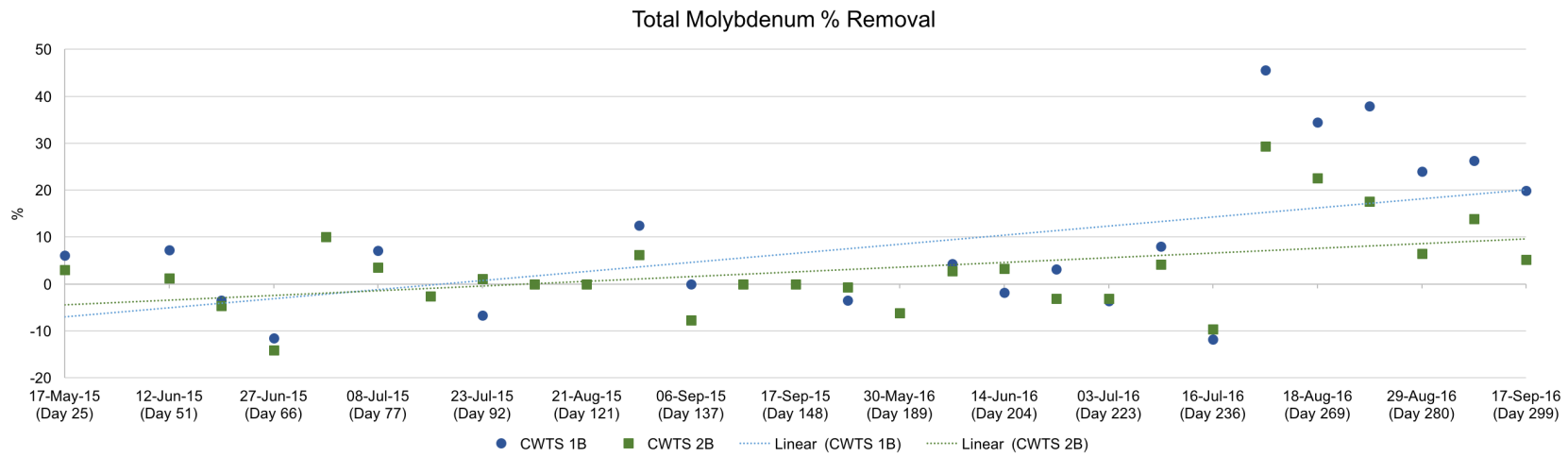


Figure B13 – Percent removal of copper in the demonstration scale wetland.
If the concentration was below the detection limit of the assay, the detection limit was used to calculate the percent removal. The Maxxam (2015 results) and The ALS (2016 results) measurement uncertainty (MU) for copper is +/- 0.0002 mg/L.



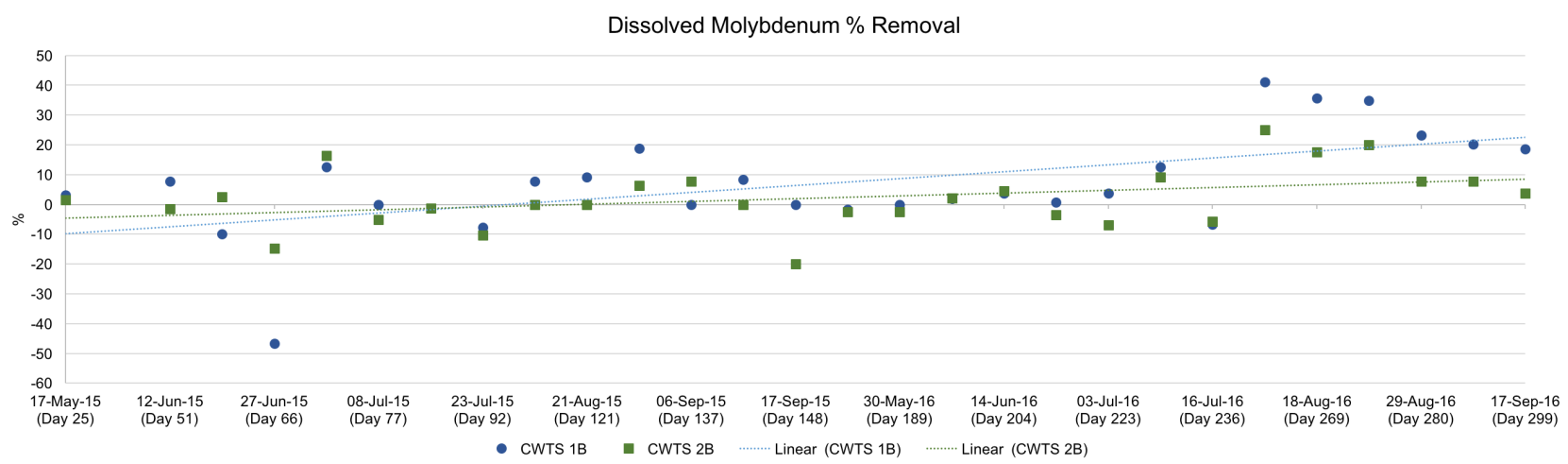
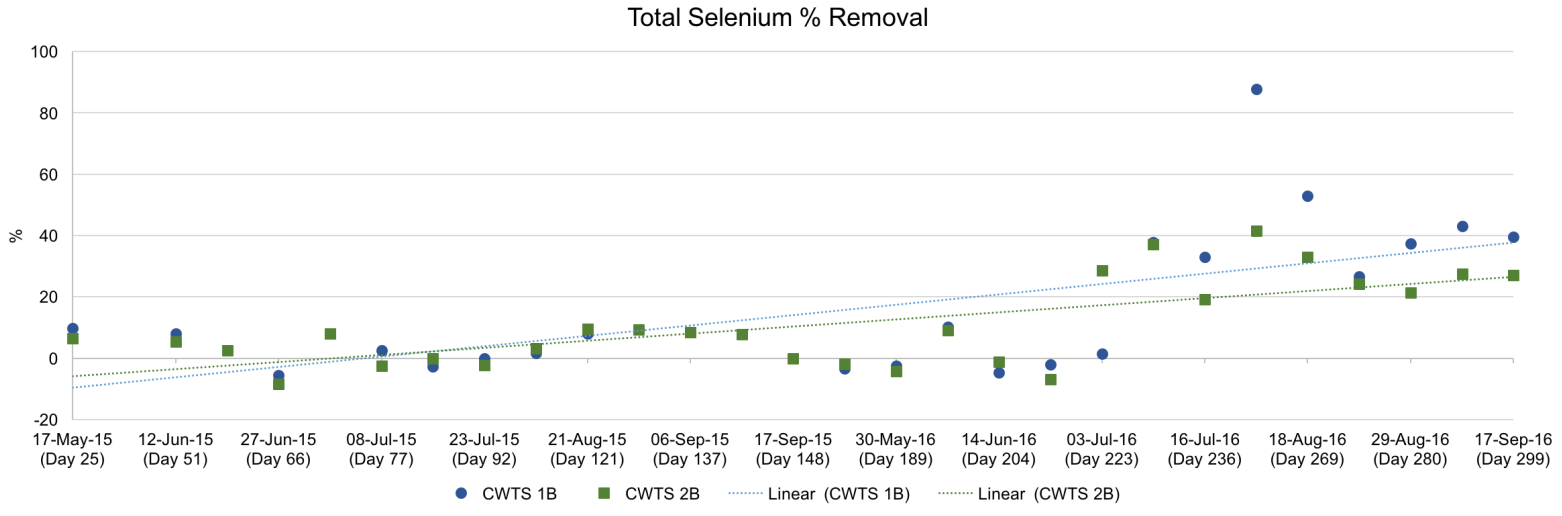


Figure B14 – Percent removal of molybdenum in the demonstration scale wetland.
If the concentration was below the detection limit of the assay, the detection limit was used to calculate the percent removal. The Maxxam (2015 results) measurement uncertainty (MU) for molybedenum is +/- 0.0002 mg/L. The ALS (2016 results) MU is +/- 0.000050 mg/L.



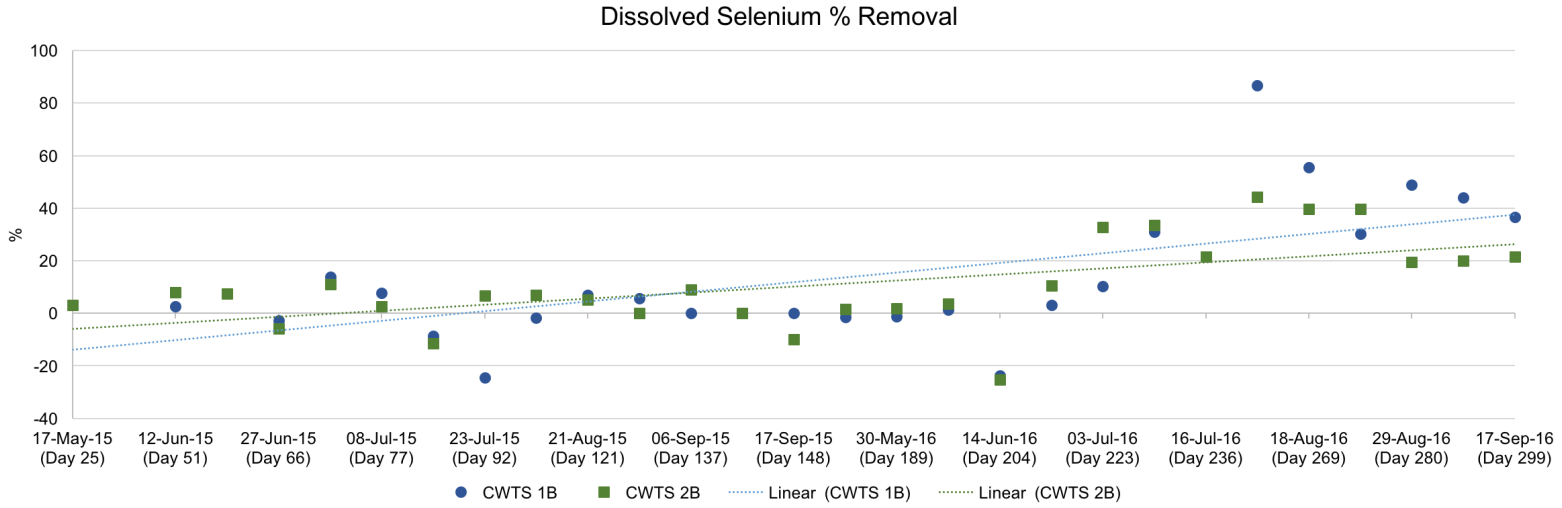
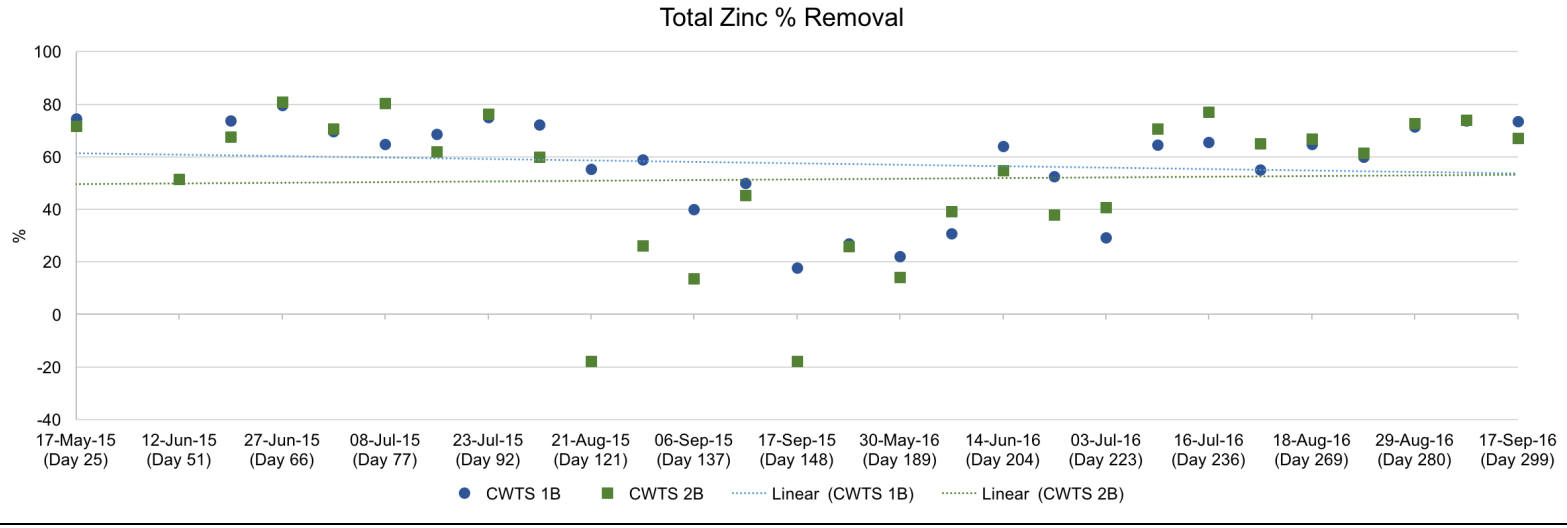


Figure B15 – Percent removal of selenium in the demonstration scale wetland.
If the concentration was below the detection limit of the assay, the detection limit was used to calculate the percent removal. The Maxxam (2015 results) measurement uncertainty (MU) for selenium is +/- 0.0002 mg/L. The ALS (2016 results) MU is +/- 0.000050 mg/L.



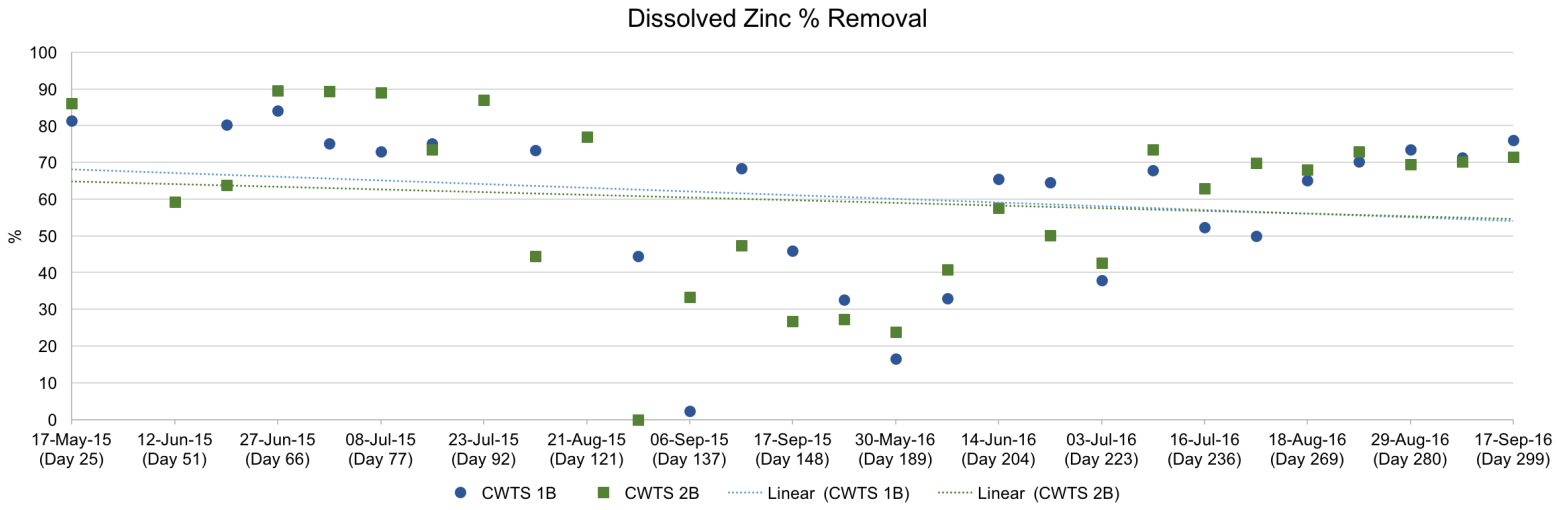


Figure B16 – Percent removal of zinc in the demonstration scale wetland.
If the concentration was below the detection limit of the assay, the detection limit was used to calculate the percent removal. The Maxxam (2015 results) measurement uncertainty (MU) for zinc is +/- 0.0030 mg/L. The ALS (2016 results) MU is +/- 0.0010 mg/L.

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Minto Mine Constructed Wetland Treatment Research Program – Demonstration-Scale 2017

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Executive Summary

The Minto Mine is an open pit and subsurface copper mine located 240 km northwest of Whitehorse in the Yukon Territory. As a result of mining activities, cadmium, copper, molybdenum, selenium, and zinc are slightly elevated beyond background concentrations and have been identified as constituents of concern in the Reclamation and Closure Plan for the Minto Mine. As part of the Reclamation and Closure Plan, a constructed wetland treatment system (CWTS) was designed by Contango with the objective of attaining suitable passive treatment of water quality at mine closure.

A phased approach is being followed to guide the site-specific design and implementation of a constructed wetland treatment system at closure. The phased approach for Minto Mine was initiated with a site assessment in 2013 that identified plants suitable for a CWTS as well as natural treatment processes that were already occurring on-site. Following the site assessment, pilot-scale testing was undertaken to confirm and provide proof-of-concept before constructing the on-site demonstration-scale wetland.

The demonstration-scale constructed wetland treatment system was constructed at Minto in 2014 and commissioned from 2015 through mid-2017 during which time the system matured and operational adjustments were made. Commissioning successfully established plant and microbial populations and achieved conditions conducive for treatment of constituents of concern. The demonstration-scale CWTS operational period ran for a month from mid-August to mid-September 2017. Various aspects of the system were measured in 2017 and are detailed in this report including operating conditions, water treatment performance, fate and distribution of treated metals, evapotranspiration, detritus decomposition rates, microbial community characterization (catalyzing treatment reactions), and pest control.

Throughout the operational period, the demonstration-scale CWTS successfully achieved an average decrease in concentrations of 0.0169 µg/L for cadmium (from 0.0261 µg/L to 0.0092 µg/L), 31.8 µg/L for copper (from 49.1 µg/L to 17.3 µg/L), 3.6 µg/L for molybdenum (from 6.3 µg/L to 2.7 µg/L), 3.5 µg/L for selenium (from 4.0 µg/L to 0.5 µg/L), and 47.3 µg/L for zinc (from 49.2 µg/L to 1.9 µg/L). Leaching of copper and other constituents from the mineralized soils used in the construction decreased by the end of 2017 with constituents showing a shift into stable reduced mineral forms in the soil (as the CWTS was designed to do for constituents from the water, but also has now done for the elements in the soil). Additional positive results were also documented; plant uptake of constituents remained minimal throughout operation and high abundance of beneficial sulphide-producing bacteria for treatment of Cu, Cd, Zn, as well as nitrate- and selenium-treating bacteria associated with plant roots.

The operational period of the demonstration CWTS confirmed that the predicted water quality for the Minto Mine at closure is amenable to treatment by these methods. Results from ongoing monitoring of the system in future years will be used to optimize performance and inform the designs for the full-scale constructed wetland treatment system.

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Abbreviations and definitions

Acid volatile sulphides (AVS) – Sulphide complexed with Fe (FeS) where the sulphide can preferentially exchange Fe for heavy metals (e.g., Ni, Zn, Cd, Pb, Cu, Hg) which are then more stably bound.

Amendment – A chemical or organic material added to encourage specific conditions (e.g., aerobic/anaerobic, pH, ORP) or as a source of something that is needed for passive treatment (e.g., nutrients, alkalinity, binding sites, etc).

Carbon source – A source of carbon (energy/electrons) for microbes (see **electron donors**). Examples include ethanol, methanol, acetate, sugar (glucose), molasses, wood chips, detritus (dead plant matter).

Carex aquatilis – A plant (emergent macrophyte) commonly known as water sedge.

Contango – Contango Strategies Ltd.

Constructed wetland treatment system (CWTS) – Wetlands that are designed and constructed to remove compounds from water, using natural processes to sequester them into the soils rendering them less bioavailable. They are different from wetlands that provide habitat for wildlife.

Constituent of Concern (COC) – Specific elements that have been identified for evaluation, including cadmium, copper, molybdenum, selenium, and zinc.

Denitrification – Where nitrate (NO_3^-) is reduced by microorganisms to form nitrite (NO_2^-), nitric oxide (NO), or nitrous oxide (N_2O), or nitrogen gas (N_2) (see **nitrate reduction**).

Dissolved oxygen (DO) – Diatomic oxygen (O_2) dissolved in water; oxygen can dissolve in water by diffusion from surrounding air, as a product of photosynthesis, or through forced aeration.

Electron donor(s) – A chemical compound that donates electrons to another compound (see **carbon source**). An electron donor is a reducing agent, which that by virtue of it donating electrons, is itself oxidized (see **oxidation** and **reduction**).

Explanatory parameters – Quantifiable parameters that indicate the type of water treatment reactions that are likely to take place (i.e., dissolved oxygen, pH, soil redox).

Evapotranspiration – The combined effects of open water evaporation and plant transpiration (Beebe *et al*, 2014).

Genetic analysis – Analysis to assess the presence, identity, and diversity of different microbes in a sample.

ICP-MS – Inductively coupled plasma mass spectrometry.

Macrophytes – An aquatic plant, large enough to be seen by eye. Can be emergent, submergent, or floating.

Microbes – Microscopic organisms that can be uni- or multi-cellular. This includes algae, bacteria, fungi, viruses, and yeast.

Most probable number (MPN) – A statistical value representing the viable population of microbes in a sample through use of dilution and multiple inoculations.

Oxidation – The loss of electrons, or increase in valence state, by a molecule, atom, or ion. Can be driven by microbes. Process is complementary to chemical reduction.

Oxidation-reduction potential (ORP) – A measure of the tendency of a chemical species to acquire or donate electrons, thus becoming reduced or oxidized, respectively, measured in millivolts.

Passive or semi-passive treatment system(s) (PTS) – General term used to refer to both passive and semi-passive treatment systems that use processes coupling transformations (e.g., chemical and biogeochemical reactions) with physical transfers (e.g., sorption, filtration) to remove constituents from water, often operationally passive with little long-term management required.

Redox – Oxidation-reduction potential (in sediment), a measure of the tendency of a chemical species acquire or donate electrons, thus becoming reduced or oxidized, measured in millivolts. This measurement is relative to the water ORP.

Reduction – The gain of electrons, or a decrease in valence state, by a molecule, atom, or ion. Can be catalyzed by microbes. Process is complementary to chemical oxidation.

Simultaneously extracted metals (SEM) – Amounts of heavy metals such as Ni, Zn, Cd, Pb, Cu, Hg in sediment, assessed in the context of AVS for excess sulphide. (also see **acid volatile sulphide**)

Sorption – The physical and/or chemical process by which one substance becomes attached to another substance.

Specific conductivity (SPC) – A measurement of electrical conductivity in water that is typically expressed in µs/cm, which has been adjusted for temperature (25°C).

Species (sp.) – One of the basic units of biological classification and a taxonomic rank. Rank in the classification of organisms below genus and above strain. Also can be used to refer to the oxidation state of a mineral (e.g., selenate and selenite are species of selenium).

SPLP – Synthetic precipitation leachate procedure.

Sulphide – An inorganic anion of sulphur that can form stable complexes with metals and make them insoluble in water (remove them from the water).

Sulphide producing bacteria (SPB) – Microbial reduction of sulphur compounds, such as sulphate, sulphite, thiosulphate, and sulphur, which produces sulphides and alkalinity. (see also **SRB**).

Sulphate reducing bacteria (SRB) – A form of sulphide producing bacteria that specifically uses sulphate for reduction (see **sulphide producing bacteria**).

Sulphide production – Microbial reduction of sulphur compounds, such as sulphate, sulphite, thiosulphate, and sulphur, which produces sulphides and alkalinity.

Thermodynamic minimum – The minimum concentration of a contaminant of concern that is consistently achievable.

Total dissolved solids (TDS) – A measure of the combined organic and inorganic salts dissolved in water.

Total organic carbon (TOC) – A measurement of the total organic carbons present in water.

Transfer – Processes that treat water by transferring a constituent to another location without changing its form. For example: absorption, adsorption, dilution, dispersion, filtration, precipitation (aqueous to solid), and volatilization.

Transform – Processes that change the chemical form or state of a constituent. For example: biodegradation, biotransformation, hydrolysis, ionization, oxidation, photolysis, and reduction.

1. Introduction

The Minto Mine, owned and operated by Capstone Mining Corp., is located 240 km northwest of Whitehorse on the west side of the Yukon River. The Minto property lies within the eastern part of the Dawson Range, with elevations from 700 to 1,000 m above mean sea level. The landscape has rounded mountains intersected by broad valleys and drainages that are part of the Yukon River watershed.

The Minto Mine has been in commercial operation since October 2007 and the deposits being mined are copper sulphide mineralized zones. Surface and groundwater water quality is a key consideration in the evaluation of potential effects of mining and mineral development projects and changes to water quality parameters have the potential to affect aquatic and human use of water resources. A Reclamation and Closure Plan (RCP) is required under both the Water License and the Quartz Mining License. The RCP is intended to address the long-term physical and chemical stability of the site and closure of the proposed features and disturbances associated with the mine. As a part of the RCP, a Constructed Wetland Treatment System (CWTS) is being designed, evaluated, and optimized for water treatment at closure through a phased program approved during the Minto Phase V/VI Expansion Project (Yukon Online Registry Project Number 2013-0100). Constituents of concern that are being evaluated through the CWTS program include cadmium, copper, molybdenum, selenium, and zinc.

1.1. Purpose and Objectives

The primary objective for operation of the demonstration-scale CWTS in 2017 was to complete commissioning and progress into operational performance. Once operational performance is achieved, the removal rate coefficients (RRC; k) can be assessed to evaluate and refine full-scale designs. To achieve the operational performance, it was recognized that the remaining copper in the soils used in construction needed to be converted into metal-sulphide form; therefore, this was addressed in the early months of 2017 and paired with evapotranspiration trials to make best use of this time.

In 2017, the monitoring program shifted focus from commissioning to testing of operational performance. Activities carried out in 2017 to achieve these objectives were:

- Monitor explanatory parameters and performance to determine when commissioning is complete and the operational period has begun;
- Assess removal of constituents from the water;
- Determine the hydraulic retention time (HRT) by tracer trial and associated correction factor to apply to the nominal (calculated) HRT;
- Evaluate CWTS performance, and determine achievable concentrations of contaminants of concern (thermodynamic minimums);
- Update site-specific removal rate coefficients (from commissioning period) with data from operational period;
- Determine amount of water loss due to evapotranspiration and effect on outflow concentrations;
- Monitor metals leaching from mineralized soils used in construction;

- Assess stability of constituents of concern in soils;
- Determine the rate and extent of detritus decomposition (*C. aquatilis* leaves) in the CWTS over time;
- Assess treatment mechanisms (including microbes); and
- Determine an appropriate method for insect pest control (aphids) in the CWTS.

1.2. Overall CWTS Project Approach

For a CWTS to be effective, it must be designed, piloted (tested), optimized, implemented, and maintained in a site-specific manner. A phased approach allows for improvements and optimization at each step. The phases used at the Minto Mine include:

1) site assessment and information gathering;
2) technology selection and conceptual design;
3) pilot-scale testing and optimization (controlled environment, off-site);
4) demonstration-scale confirmation and optimization (on-site); and
5) full-scale implementation.

Phases 1 to 3 are complete at Minto. This work is summarized in the Research Program, Pilot Plant and Demonstration reports completed by Contango (2014a and 2014b). Phase 4 of the project is underway, with the on-site demonstration CWTS constructed at the Minto Mine during fall 2014 (Contango, 2015). Commissioning of the CWTS occurred from 2015-2017 with operations beginning in late 2017. Performance results of the demonstration CWTS indicate that the CWTS is maturing as expected (Contango, 2015; 2016; 2017). This document reports on the 2017 on site demonstration-scale CWTS data, with focus on results from the operational period from August 18 – September 22, 2017.

2. Demonstration-Scale Constructed Wetland Treatment System Design

2.1. CWTS Layout and Dimensions

The demonstration-scale CWTS is located on the northeast side of the mine site, perched on the MVFE area as shown in Figure 1.

Construction of the demonstration-scale CWTS was completed in 2014 and includes two series in parallel with two cells in each series and a final catchment basin (Figure 1). Water flows from the feed tank through the A cells, through the B cells and into the final combined catchment basin. Water from the catchment basin of the demonstration-scale CWTS is not discharged off-site, rather it is collected in a sump at the toe of the Mill Valley Fill Extension (MVFE). Series 1 and series 2 flow independently of each other and are intended as replicates for analytical testing and operation confirmation. Additional details can be found in the Minto Demonstration-Scale Report (Contango, 2015) and Appendix A of this report.



Figure 1 – Picture of demonstration-scale CWTS at Minto Mine.
Demonstration-scale CWTS location in relation to its surroundings. To the far right is the mine camp while below is the tree line. This photo was taken on September 24, 2017.

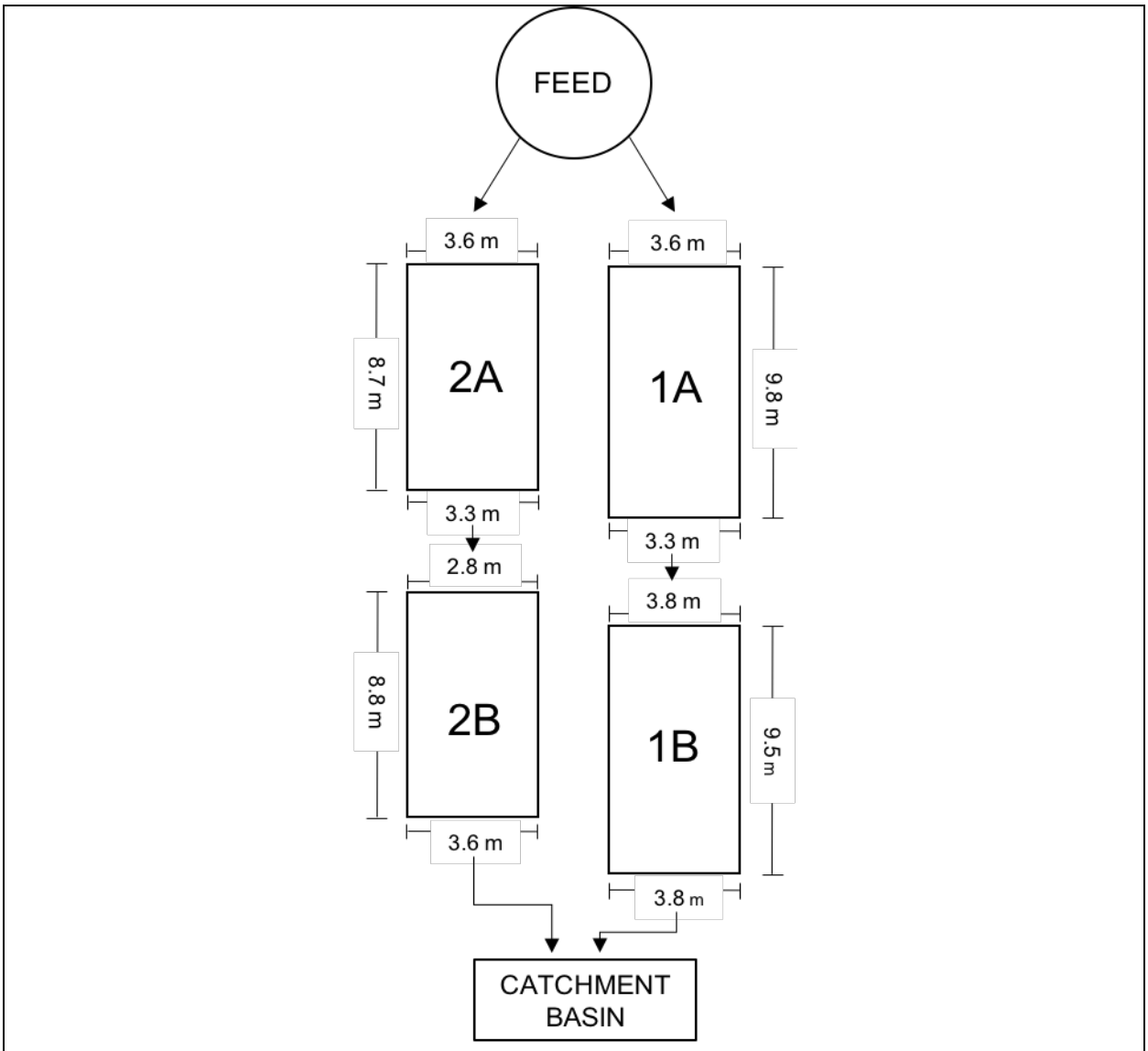


Figure 2 – Diagram of demonstration-scale CWTS.
Dimension measurements are indicated at soil surface. Water flows are indicated with black arrows. Water flows from the feed tank into the A cells, into the B cells then into the final catchment basin. Series 1 and series 2 flow independently of each other, and serve as replicates.

2.2. Soils

Soils used to construct the CWTS are described in Appendix A of this report and in the initial report that outlines construction (Contango, 2015). In brief, the recommended soil for the CWTS was sand, with 2-7% by volume organic material (e.g., woodchips, peat). The material used in the construction of the demonstration-scale CWTS was an organic peat, and analyses received after construction indicated an elevated concentration of leachable copper (Appendix A, Table A2). This leachable copper has affected the CWTS performance results. Therefore, this has been a focus of efforts through the commissioning period to transition these leachable copper minerals into stable sulphide minerals which will not leach from the CWTS.

2.3. Vegetation

The demonstration-scale CWTS was planted with *Carex aquatilis* (aquatic sedge) and aquatic mosses (bryophytes) from the W10 area of the Minto Site. The plant selection and borrow source were identified during the site assessment in 2014 (Contango, 2014a) and pilot-scale testing (Contango, 2014b).

2.4. Water Source

The water source for the demonstration-scale CWTS is seepage from the toe of the Mill Valley Fill Extension (MVFE) and was selected for the demonstration-scale CWTS as the seepage is similar to that expected upon closure in the MVFE area. Zinc concentrations were elevated in the water being used in the CWTS in 2016, and it was uncertain if this was due to a culvert used for water collection in the area or seasonal variation. However, zinc concentrations remained elevated in 2017, suggesting that the culvert is contributing to elevated zinc concentrations. Further details can be found in Appendix A and Contango's 2016 update report (Contango, 2017).

3. Commissioning and Operations

The period between the construction of the CWTS and achieving the expected treatment performance is referred to as the commissioning period. During this period, several criteria are monitored to determine when the commissioning period has been successfully completed. Criteria that were achieved that indicated the end of the commissioning period and the beginning of the operational period for the Minto Mine demonstration-scale CWTS include the following and are further discussed in the sections listed:

- Plant establishment and maturation such that the plants have grown in to densities visually similar to natural wetlands in the area, but in monoculture (Section 3.2),
- Establishment of reducing conditions within the CWTS (i.e., average soil redox was below -100 mV consistently in 2017; Section 5.1),
- No increase in aqueous copper concentrations through the CWTS (due to copper containing soils used in construction;Sections 5.2.1 and 5.6), and
- Microbial population establishment and maturation to levels similar or better to pilot-scale (Section 5.9).

Once the above criteria were met, the CWTS was deemed to be in the operational period. The same criteria, other than the copper leaching, could be used for the full-scale CWTS. The on-site demonstration-scale CWTS was commissioned from 2015 to 2017, comprising of 135 days in 2015, 150 days in 2016 and 82 days in 2017 (Table 1). The CWTS then proceeded to the operational period and operated for 35 days in 2017 (Table 1).

Table 1 – Days of operation of demonstration-scale CWTS.

Scale	Year	Period (days)	Date	
			Start	End
Commissioning-A[2]	**2014[1]**	23	Aug 27	Sept 19
	2015	135	May 16	Sept 29
	2016	87	May 2	Jul 28
Commissioning-B[3]	**2016**	63	Jul 29	Sept 30
	2017	82	May 27	Aug 17
Operational Period	**2017**	35	Aug 18	Sept 22

[1] The CWTS was constructed in 2014, but no water testing occurred during this first month of commissioning-A.
[2] The end of the commissioning-A period and the beginning of the commissioning-B period was marked by the addition of organics on July 28, 2016.
[3] The end of the commissioning-B period and the beginning of the operational period was marked by stabilization of flow rates and resolution of feed water delivery complications.

For the demonstration-scale CWTS the commissioning period was divided into two periods, commissioning-A, and commissioning-B. The end of the commissioning-A period and the beginning of the commissioning-B period was marked by the addition of organics on July 28, 2016. These organics (straw and wood chips) were added to further aid the copper in the soils to transition into sulphide mineral forms, and represents an amount of organic material similar to what would be produced by the CWTS once fully established with *Carex aquatilis*.

The end of the commissioning-B period was expected to be between July 16 and August 2, 2017. However, due to issues with feed-water pumps and flow meters, the end of the commissioning-B period was extended to August 17, 2017 and was marked by stabilization of flow rates and resolution of feed water delivery complications. Additionally, sandbags were added at the end of each cell on August 11, 2017 to increase the water depth and further promote reducing conditions. Therefore, the operational period began on August 18, 2017 and ran until September 22, 2017. Flow was turned off to prepare for winter freeze-up on September 30, 2017, and the last sampling date used was September 22, 2017. Discussions in this document are focused on the operational period and further details on commissioning in 2015 and 2016 can be found in Contango's past reports (2015 and 2016).

3.1. Flow Rates

The targeted flow rates were varied during commissioning of the demonstration-scale CWTS to target desirable conditions for establishment of the CWTS. Since the cells in series 2 are smaller than cells in series 1, a faster flow rate is used for series 2 to obtain an equal HRT to series 1.

In 2017, a long HRT was selected to aid in generating reducing conditions in the CWTS and, therefore, flow rates were set to as slow as they could operate. The nominal HRT was targeted to be 5 days. The operational ranges of the flow meters used in the CWTS were 0.3 to 3 gallons per minute (GPM). To obtain this long HRT, the targeted flow rates used during the commissioning and operational periods of the CWTS in 2017 were 0.37 and 0.31 GPM for series 1 and series 2, respectively. Due to pump and flow meter issues, the actual flow rates varied from the target flow rates throughout 2017 commissioning and operational periods. Table 2 summarizes known pump and flow meter issues that occurred in 2017. The average flow rate for the 2017 operational period was a calculated flow rate of 0.38 GPM and 0.29 GPM for series 1 and series 2, respectively. Additional information on flow rates and associated HRT calculations are provided in Appendix A.

3.2. Health and Establishment of CWTS Vegetation

3.2.1. Carex aquatilis

During the commissioning period, plants establish and mature, with density expected to increase over time. From planting in 2014 to the last site visit in 2017 *C. aquatilis* thrived in the CWTS creating a dense emergent macrophyte monoculture, supplemented with aquatic mosses (Figure 3). This suggests that *C. aquatilis* are very robust and reaffirms that they are a good candidate for use in the full-scale CWTS at Minto. In July 2017, an aphid infestation occurred in the CWTS which affected the *C. aquatilis*. The biomass above water appeared to partially die off, however the plants continued to send out new shoots suggesting the *C. aquatilis* were resilient to the infestation. Further discussion about the aphids and the control measures in 2017 is presented in Section 6.

3.2.2. Moss

Aquatic mosses have continued to mature and expand in size from 2014 through 2017. The mature mosses are beginning to show characteristics of the desired coupled transfer (sorption, filtration) and transformation (mineralization, reduction) processes. The top of the moss is growing and producing new green biomass that provides transfer sites, and the older, bottom of the moss is turning black and beginning to decompose which creates sulphide reducing zones that allow for transformation (Figure 4 and Figure 5).

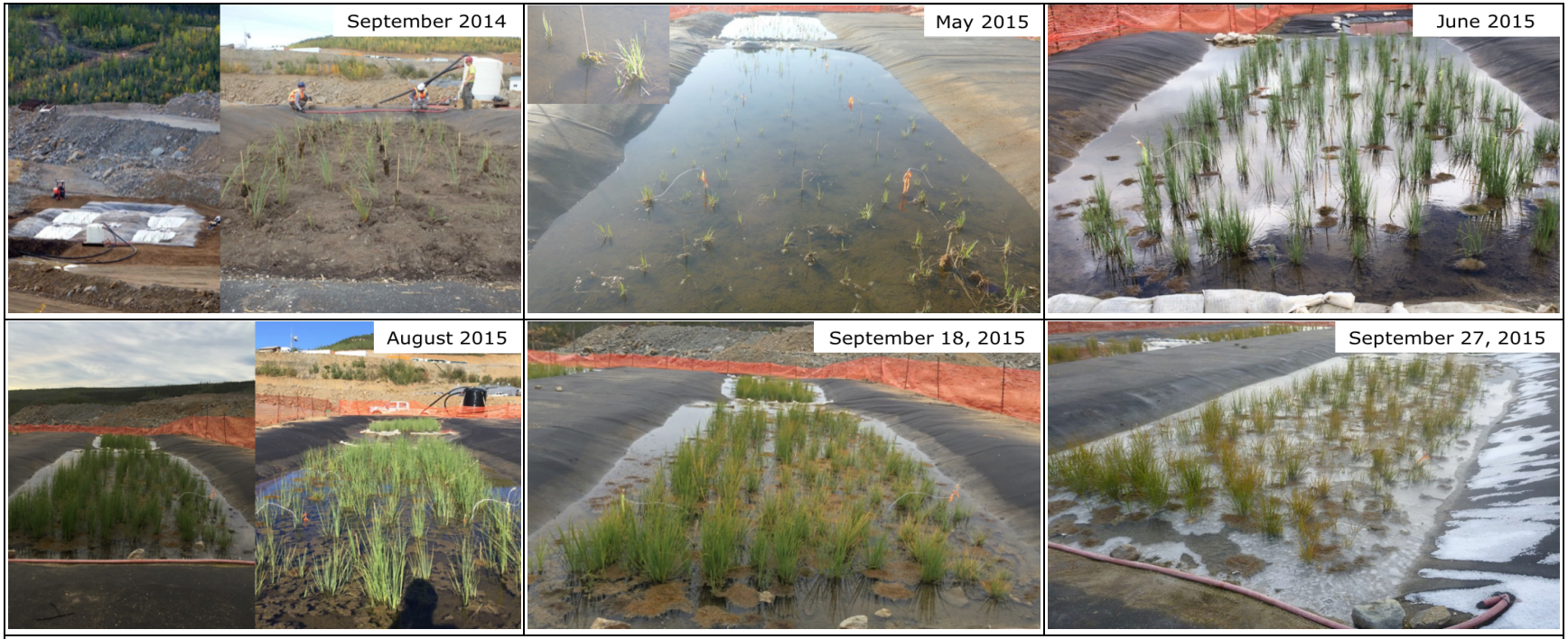


Figure 3 – Maturation of the CWTS from construction through operations.
2015 pictures show cell 2A.

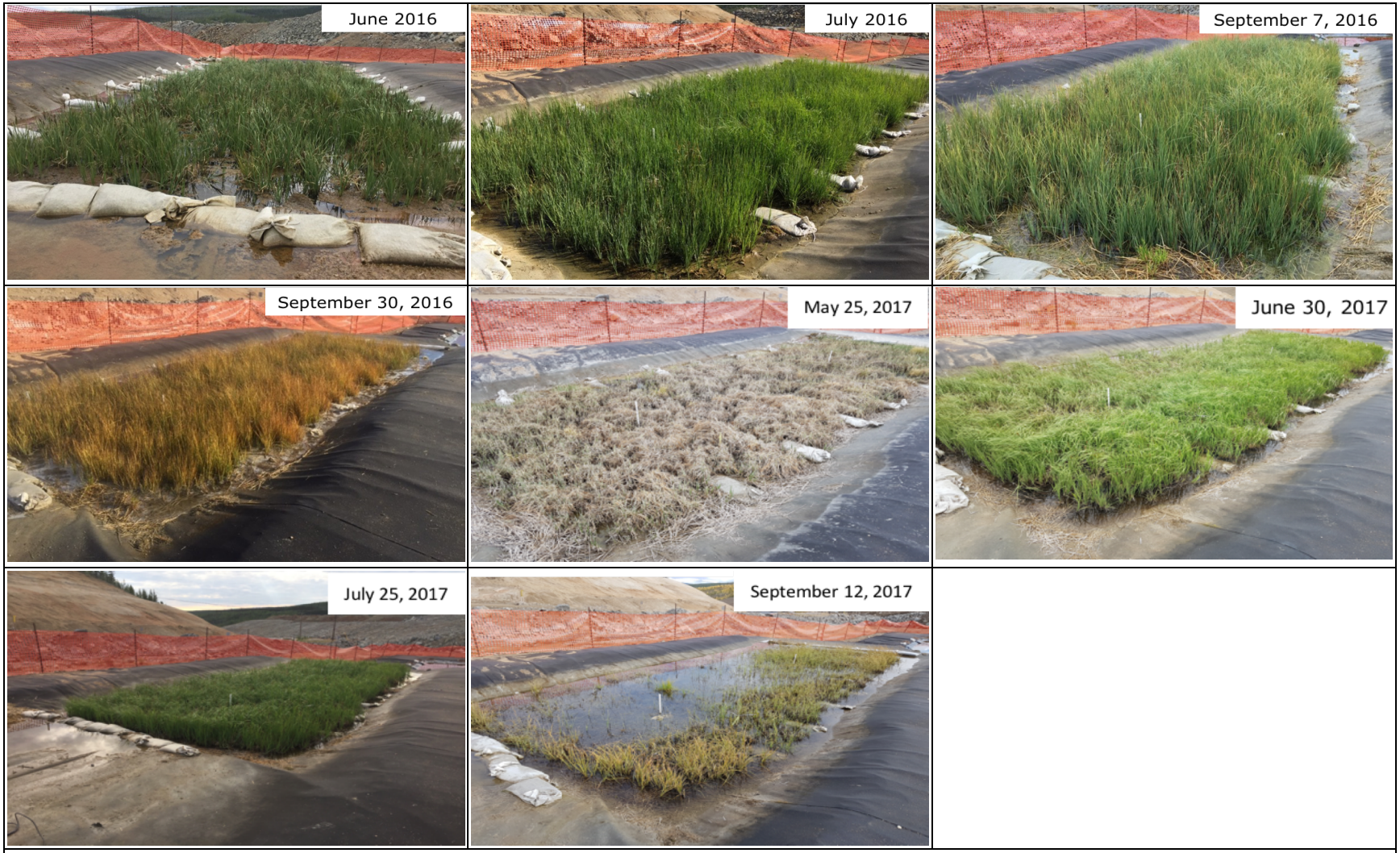


Figure 3 continued – Maturation of the CWTS from construction through operations.
2016 and 2017 pictures show cell 1B. September 12, 2017 shows increased yellowing and die off in cell 1B due to aphid infestation.



Figure 4 – Moss maturation in CWTS through 2016.



Figure 5 – Moss top and bottom view in June 2016 and September 2017.

4. Timeline and Sampling Schedule

Major events and operational adjustments or amendments for 2017 are listed in Table 2. Details for events up to 2017 can be found in the Minto Demonstration-scale 2016 Update Report (Contango, 2017). The actual dates as well as effective days of operation are provided, which adjusts for time that the CWTS was not receiving water when it was frozen. The effective days of operation allows for comparison to expected timelines from the pilot-scale testing, and for planning and scheduling to be done for full-scale construction and commissioning. The sampling schedule for 2017 was conceptually developed prior to beginning construction of the demonstration-scale CWTS (Appendix A). Actual dates of sampling were adjusted for timing of spring thaw and winter freeze-up, and the associated ability to have the pumps operating at the W62 sump to supply water to the demonstration-scale CWTS.

Table 2 – Events and sampling activities since construction.

Event	Key Activity	Flow Rate Setting m^3/day (gal/min)		Dates	Effective Days of Operation
		CWTS Series 1	CWTS Series 2		
CWTS constructed and planted	First sampling, water started.	*	*	August 27 – 31, 2014	0-4
Freeze up for winter	Feed water pumps turned off.	*	*	September 19, 2014	23
Winter 2014/2015					
Commissioning-A[2]				May 16 – September 29, 2015	136
Winter 2015/2016					
Commissioning-A[2]				May 2 – July 27, 2016	86
Commissioning-B[2]				July 28 – September 30, 2016	64
Winter 2016/2017					
Flow started	Flow started.	2.18 (0.40)	1.85 (0.34)	May 27, 2017	312
Evaporation Study	Flow stopped.	-	-	June 8, 2017	324
End of Evaporation Study	Flow re-started.	4.80 (0.88)	4.58 (0.84)	June 15, 2017	331
Flush CWTS	Flow increased.	5.07 (0.93)	4.20 (0.77)	June 15 – 19, 2017	331-335
Contango Site Visit #7 (Spring Sampling)	Microbiology, soils, and water tested.	2.07 (0.38)	1.42 (0.26)	June 20-22, 2017	336-338
	Detritus decomposition trial started.			June 21, 2017	337
Saline Tracer Study	Salt added to cells for tracer study.	1.74 (0.32)	0.55 (0.10)	June 21 – July 2, 2017	337-348
Flow interruptions	Flow rates increased causing feed tank to collapse, and flow stoppage.	*	*	July 16, 2017	362
Contango Site Visit #8 (Summer Sampling)	Microbiology and soils tested. Detritus bags collected.[3]	*	*	July 25 – 26, 2017	371-372
Flow interruptions	Pump issues. Repairs required and new totalizer installed on CWTS 1.	*	*	July 25 - August 16, 2017[1]	371-393
Sprayed for aphids	Insecticide applied 7 times to CWTS.	N/A	N/A	July 25 – September 13, 2017	371-421
Water level raised	Sand bags were added to the ends of each cell.	*	*	August 11, 2017	388
Beginning of operational period				August 18, 2017	395
Summer Water Sampling	Not completed during Contango Site Visit #8 due to flow interruptions.	2.40 (0.44)	1.64 (0.30)		
Fall Sampling	Completed by Minto. Microbiology, soils, water, and plants tested. Detritus bags collected.	2.02 (0.37)	1.42 (0.26)	September 11, 2017	419
Freeze up for winter	Flow stopped.	-	-	September 30, 2017	438

* This indicates no flow rate was given, flow rate was unmeasurable or variable; however, cells remained flooded throughout.
[1] Resolution of the flow interruptions marked the end of commissioning-B and the beginning of the operational period.
[2] Detailed information for Commissioning-A and Commissioning-B in 2015 and 2016 can be found in the Minto Demonstration-Scale 2016 Update Report (Contango, 2017).
[3] Due to flow interruptions during the Contango site visit, the summer seasonal water sampling occurred on August 18, 2017 when flows had re-started.

5. Performance

5.1. Monitoring Explanatory Parameters

The following are key findings regarding explanatory parameters, which are detailed further below:

- Dissolved oxygen (DO) decreased from an average of 8.4 mg/L during commissioning-B in 2016, to an average of 5.3 mg/L during operations in 2017. This elevated DO in the water column is likely the result of photosynthesis of algae and mosses. Despite this DO level in the water column being in oxidizing ranges, stable reducing conditions were achieved in CWTS soils within the targeted soil redox range (-100 to -250 mV).

Average water temperature of the demonstration-scale CWTS in 2017 were similar to those in 2015 (12.9°C) and 2016 (10.2°C). The average water temperature during the operational period in 2017 was 9.7°C, ranging from 5.3°C to 14.9°C. As expected, conductivity did not change from previous years and pH remained circumneutral.

DO concentrations in the CWTS cells were on average 5.3 mg/L, which is lower than commissioning-A and slightly higher than commissioning–B (Table 3). Reducing conditions are needed for nitrate and selenium treatment processes, and for creating metal sulphides that remove copper and cadmium from the water. Water oxidation-reduction potential (ORP) also decreased compared to 2016 by 73 mV, which is also indicative of reducing conditions in the CWTS.

Table 3 – Average in situ measurements from the pilot scale and demonstration-scale CWTS.

Testing Period		DO (mg/L)	Conductivity (µS/cm)	pH	ORP (mV)	Soil redox (mV)
Demonstration-scale 2015		10.0	817.9	8.11	147.9	-52
Demonstration-scale 2016	Commissioning-A[1]	15.9	890.9	7.79	143.7	-85
	Commissioning-B[2]	8.4	1020	7.59	157.6	-89
Demonstration-scale 2017	Commissioning-B[3]	4.3	795.9	7.43	18.7	-162
	Operational period[4]	5.3	879.7	7.36	124.9	-152

[1] Data for commissioning-A period is from May 2-July 28, 2016.
[2] Data for commissioning-B period is from July 29-September 30, 2016.
[3] Data for commissioning-B period is from May 27-August 17, 2017.
[4] Data for operational period is from August 18-September 22, 2017.

A key explanatory parameter used to monitor maturation of the CWTS during the commissioning period is the soil redox potential, which is measured using platinum tip probes (in soil) and Calomel electrodes (in water). This measurement offers insight into the direction of electron flux between the sediment/soil/pore water and overlying water column (Faulkner et al., 1989; Huddleston & Rodgers, 2008), and can be used to confirm reducing conditions

in the soil. Based on the information gathered in pilot-scale testing, the targeted soil redox for the demonstration-scale CWTS is between -100 and -250 mV. In these redox ranges, bacterial sulphide-production through reduction of sulphur compounds (e.g., sulphate) is expected. Sulphide production directly results in metals and metalloid treatment for constituents such as cadmium, copper, molybdenum, nickel, lead, and zinc by precipitation as metal sulphides (Figure 6).

This maturation period is necessary for sufficient quantities of microbes to populate the CWTS and become active in decomposing organic material. It is the electrons produced by the decomposition of organic material that is reported by the soil redox measurements. The decomposition of organic material then feeds the sulphate-reducing bacteria the type of energy they need to produce the sulphides that remove the copper, cadmium, molybdenum, and zinc from the water. The microbial activity of the CWTS is discussed further in Section 5.3.2.

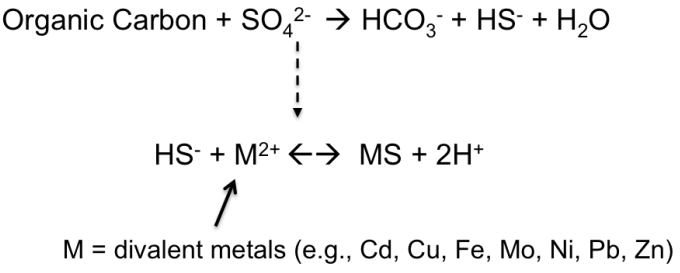


Figure 6 – Example equation of sulphate reduction and resultant metals mineralization and treatment.
Equation is not balanced; demonstrates sulphate reduction with creation of alkalinity and sulphides, which can then form metal sulphide complexes.

As expected from the pilot-scale testing, the soil redox in all the demonstration-scale CWTS cells has decreased and stabilized over time, indicating maturation of the CWTS (Figure 7). At the end of 2016, the demonstration-scale CWTS had begun achieving soil redox values that are conducive to sulphide production due to the decomposition of the organics that were added on July 28, 2016 (Contango, 2017). By the end of 2017, soil redox values had decreased to within the targeted range, even without additional organics added to the CWTS. It is therefore evident that the commissioning period was successful and the CWTS has matured and become self-sufficient in producing organic matter to provide electrons to generate reducing conditions upon decomposition. Soil redox will continue to be monitored throughout operation in 2018.

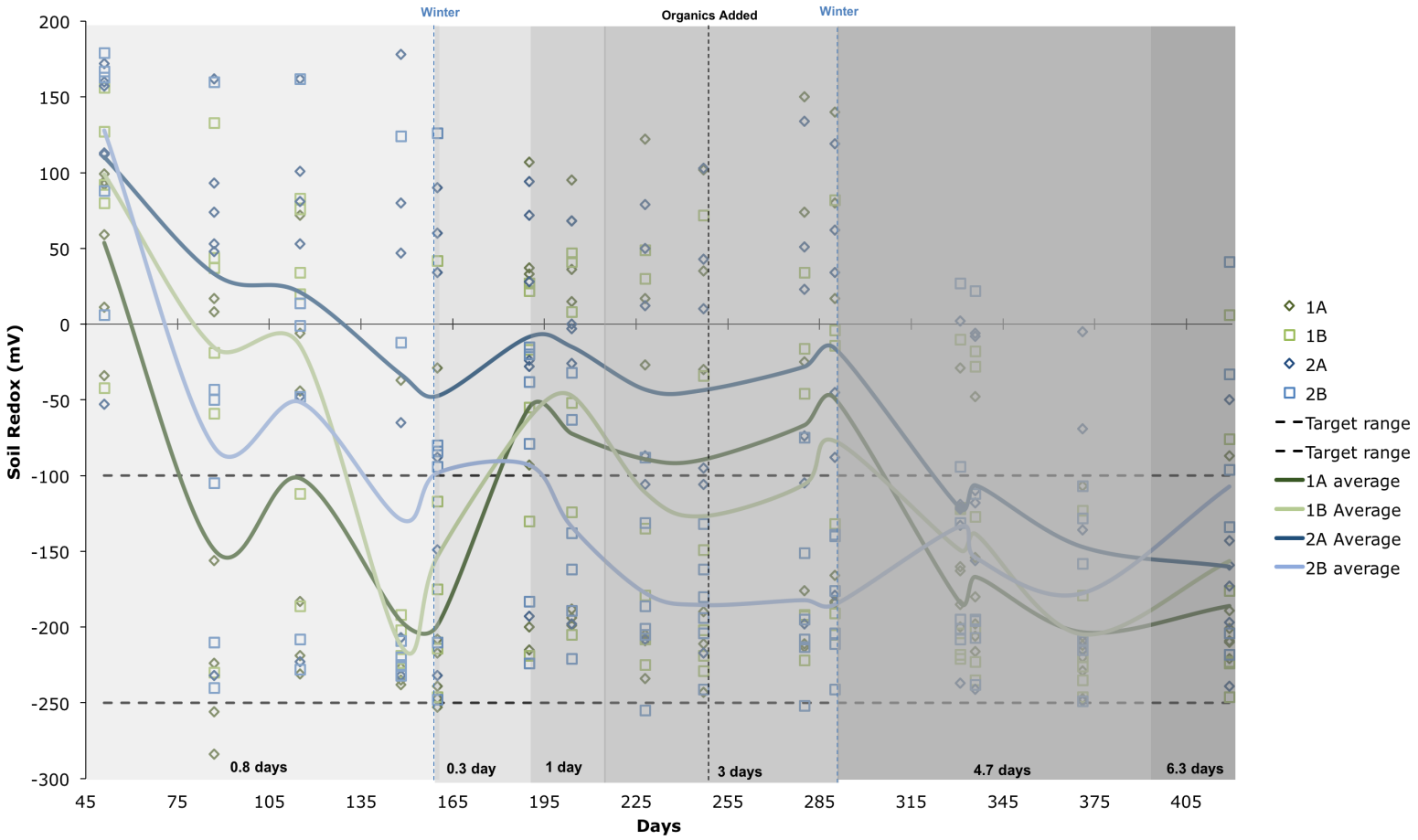


Figure 7 – Soil redox potential of each CWTS cell over time.
All demonstration-scale CWTS soil redox measurements are plotted. Targeted soil redox values based on pilot-scale testing are indicated with dotted lines. The blue dotted line indicates break in measurements for winter 2015 and 2016. Days and associated grey shading in 2015 and 2016 indicate the nominal HRT. Days and associated grey shading in 2017 indicate the actual HRT (average of series 1 and 2) using recorded water depths and flow rates during those time periods.

5.2. Water

The following are key findings regarding constituents in the water, which are detailed further below:

- Copper treatment in the CWTS was masked by leaching from the soils used in construction of the CWTS into the water, but this has mostly been remedied now by the wetland treating this copper and turning it into more stable sulphide forms in the soil.
- During the operational period the demonstration-scale CWTS successfully achieved an average decrease in concentrations of 0.0169 µg/L for cadmium (from 0.0261 µg/L to 0.0092 µg/L), 31.8 µg/L for copper (from 49.1 µg/L to 17.3 µg/L), 3.6 µg/L for molybdenum (from 6.3 µg/L to 2.7 µg/L), 3.5 µg/L for selenium (from 4.0 µg/L to 0.5 µg/L), and 47.3 µg/L for zinc (from 49.2 µg/L to 1.9 µg/L).
- Molybdenum and selenium treatment in the operational period is notable as the removal rates were negligible within the margins of error of the testing method in the commissioning-A period.

5.2.1. Metal Leaching from Soils into Overlaying Water

Although treatment improved through commissioning as the CWTS matured, copper and aluminum leaching from the soils into the overlaying water, masking the effects of treatment (Contango, 2017). Therefore, additional sampling was performed through the CWTS to identify these fluctuations. Details of the sampling methods can be found in Appendix A of this document and Figure A4 in Appendix A shows the sampling locations in series 2 of the demonstration-scale CWTS. Graphs showing concentrations of copper and selenium throughout the CWTS are presented below (Figure 8 and Figure 9, respectively) while graphs for the remaining constituents can be found in Appendix B (Figures B1 to B4).

Since the beginning of operations of the demonstration-scale CWTS, significant concentrations of copper have been leaching from the CWTS soils used in construction (Contango, 2016 and 2017; Section 5.6). Moreover, aluminum (which could not be accounted for by the influent water chemistry) was elevated in the CWTS. Because of metals leaching from the soils, the treatment occurring within the CWTS was far greater than what was being observed by simply measuring the inflow and outflow points (Contango, 2017). These soils first needed to be treated, before significant treatment could occur for the influent waters.

In early 2017 (June 20, 2017 sampling date), copper was still leaching from the soils into the water (Figure 8). However, at later sampling dates in 2017 copper was no longer leaching at a rate that the treatment wetland could not keep up with for treatment. Copper concentrations decreased by the end of the A cells and stabilized through B cells, indicating that treatment of copper is occurring and that copper leaching from the soils has subsided. Leaching of constituents from the soils is further described in Section 5.6.

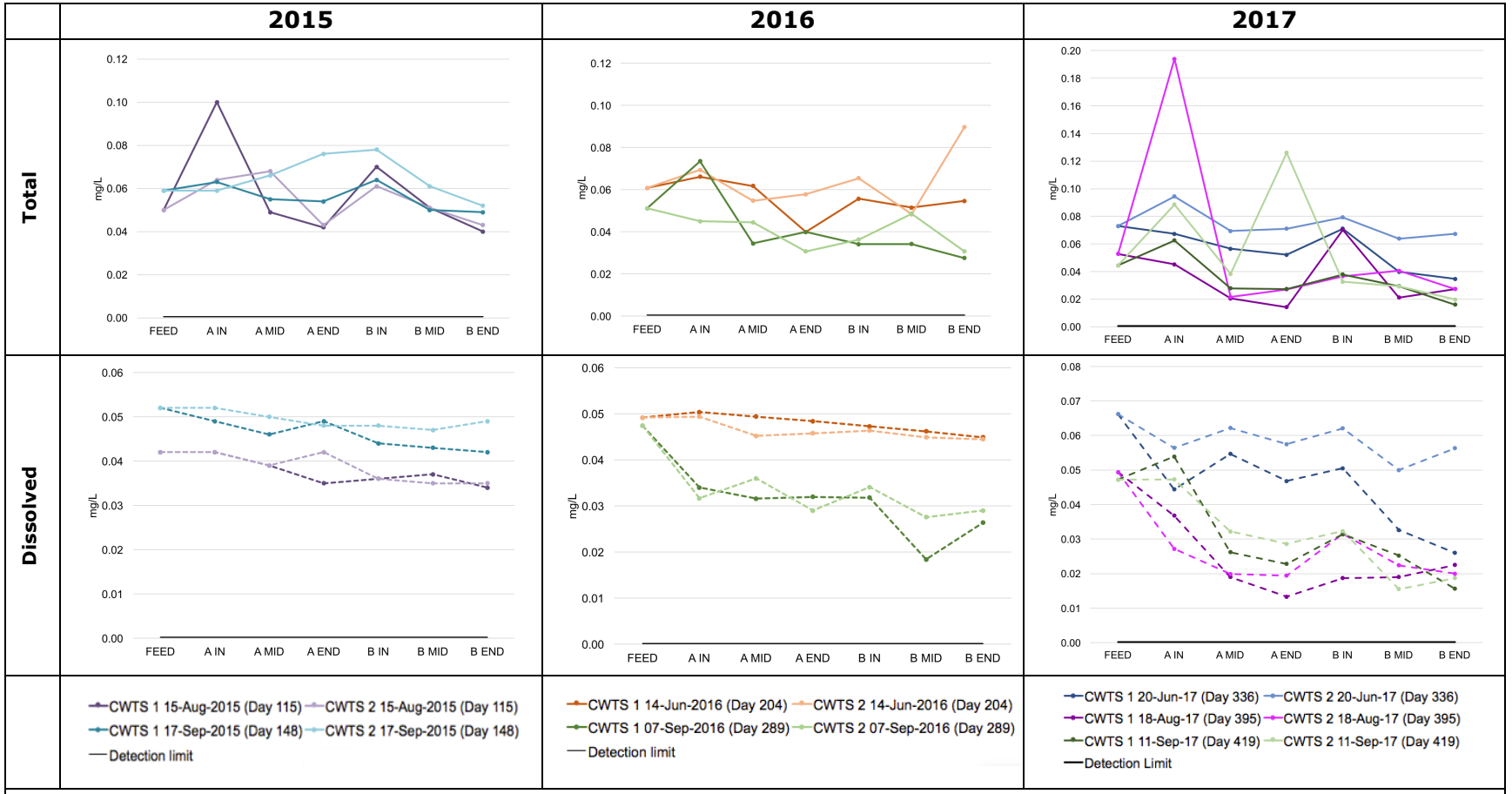


Figure 8 – Copper concentrations through the CWTS.
2015 (left), 2016 (middle), and 2017 (right) total (top) and dissolved (bottom) copper concentrations. Data shown for seven timepoints, where water was sampled at 7 locations through the flow path of the CWTS to assess for treatment fronts within the CWTS, or possible leaching of constituents from the soils into the CWTS. The Maxxam (2015 results) detection limit (DL; black line) for copper is 0.0002 mg/L. The ALS (2016 and 2017 results) DL for copper is 0.0005 mg/L. Spikes in copper concentrations were observed in early sampling points in the CWTS indicating leaching from the soils is occurring; however, later sampling points do not show any copper spikes and, therefore, leaching has subsided. The 2017 graphs are on different y-axes than previous graphs due to one data point from 18-Aug-17 for total copper and two data points from 20-Jun-17 for dissolved copper that are higher than previous axes.

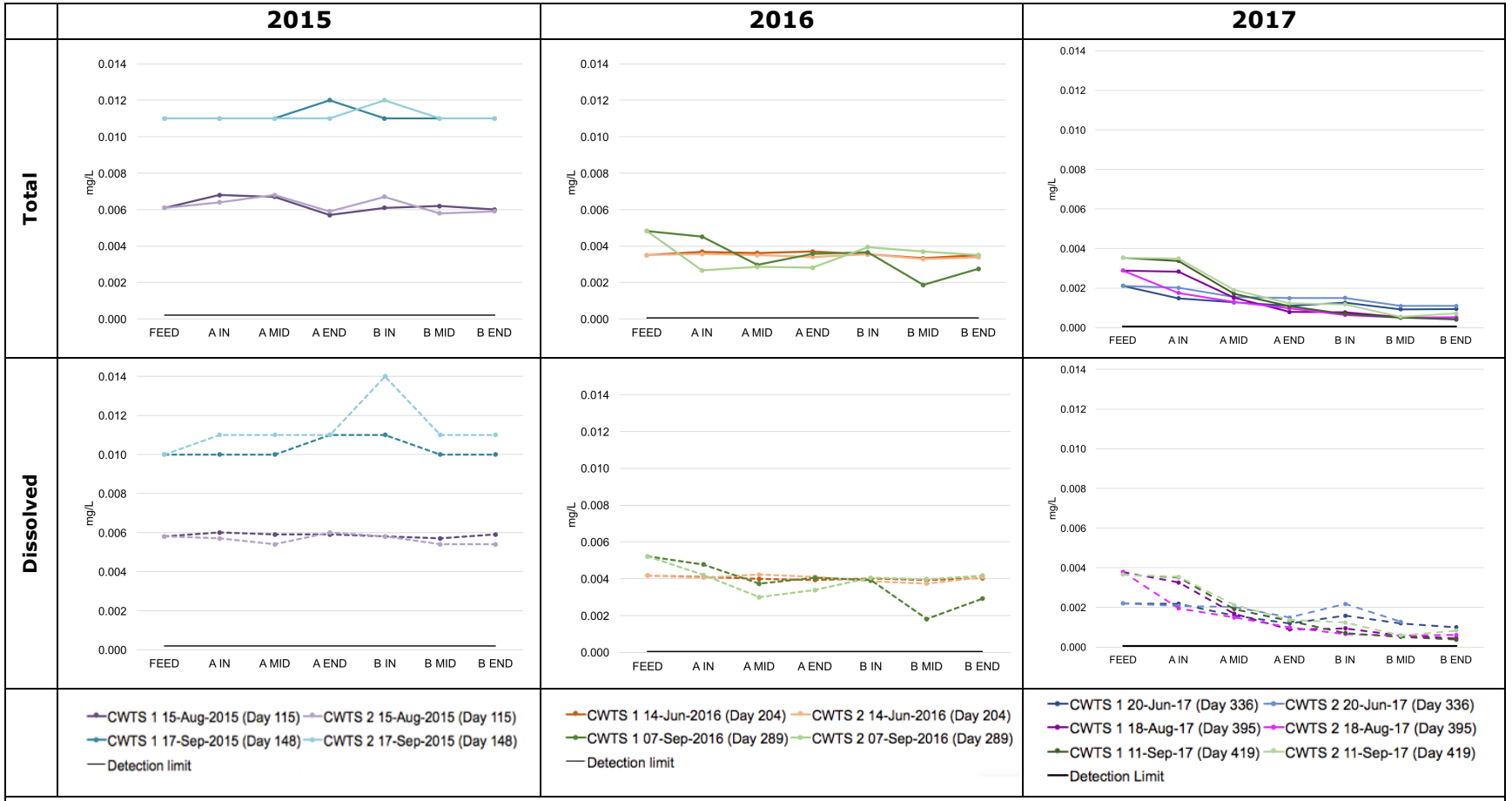


Figure 9 – Selenium concentrations through the CWTS.
2015 (left), 2016 (middle), and 2017 (right) total (top) and dissolved (bottom) copper concentrations. Data shown for seven timepoints, where water was sampled at 7 locations through the flow path of the CWTS to assess for treatment fronts within the CWTS, or possible leaching of constituents from the soils into the CWTS. Y-axes are set to be the same for total and dissolved. The Maxxam (2015 results) detection limit (DL; black line) for selenium is 0.0002 mg/L. The ALS (2016 and 2017 results) DL is 0.000050 mg/L. This graph indicates that leaching is not occurring as there are no spikes in selenium concentrations throughout the CWTS.

5.2.2. Performance during Commissioning-A

The commissioning-A period occurred from May 16 to September 29 in 2015 and May 2 to July 28 in 2016. Details regarding the results of the commissioning-A period can be found in Demonstration-Scale 2016 Update Report (Contango, 2017).

5.2.3. Performance during Commissioning-B

The beginning of commissioning-B was marked by the addition of organics to the CWTS and ran from July 29 to September 30 in 2016 and from May 27 to August 17 in 2017. These organics (straw and wood chips) were added to further aid the copper in the soils to transition into sulphide mineral forms, and represents an amount of organic material similar to what would be produced by the CWTS once fully established with *Carex*. Treatment of constituents was similar through 2016 and 2017 for commissioning-B. An evapotranspiration trial was also completed during commissioning-B to allow water to stagnate and provide time for copper in the soils to convert to sulphide mineral forms more readily.

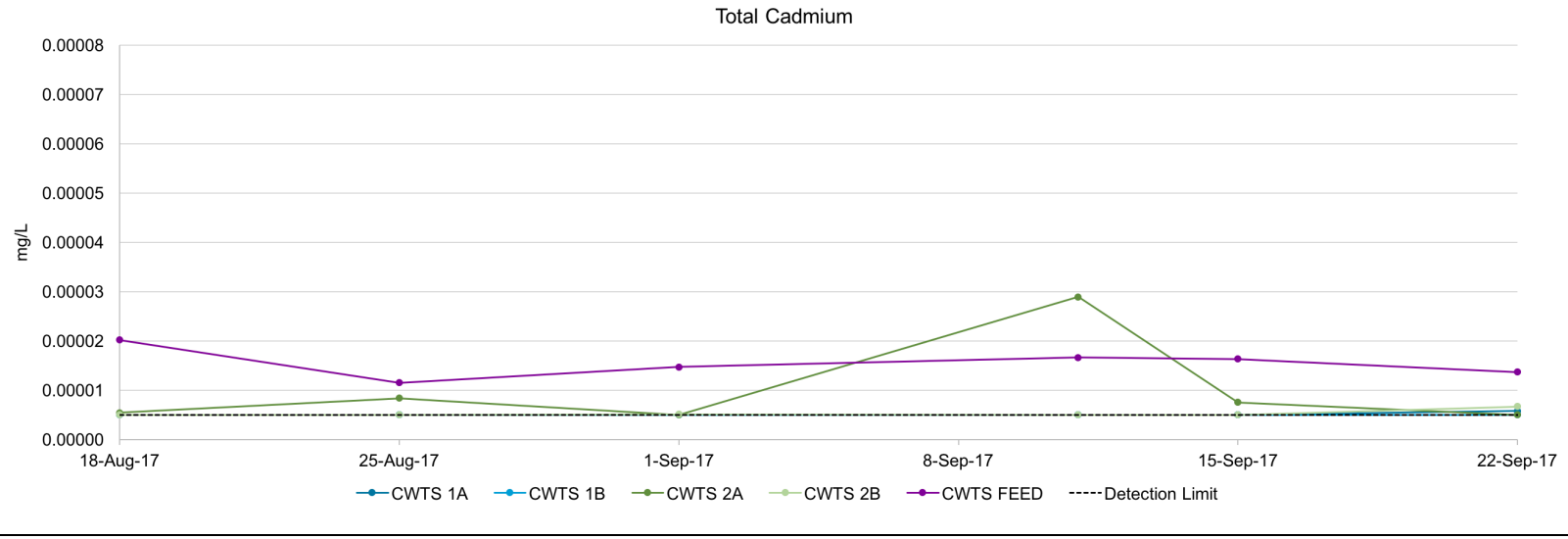
5.2.4. Performance during Operations

Figures showing performance of the demonstration-scale CWTS in 2017 (commissioning-B and operational period) can be found in Appendix B (Figures B5 to B12). In the operational period from August 18 to September 22, 2017, dissolved cadmium, copper, selenium, and zinc treatment improved from the 2017 commissioning-B period. Dissolved metals concentrations were used for the discussion in this report instead of total metals concentrations, as total values were highly variable and not representative of the metals concentrations in the CWTS. The variability of the total metals concentrations is likely owing to particulate material containing metals being part of the grab sample collection method. During the operational period the CWTS achieved an average decrease of 0.0169 µg/L for cadmium (from 0.0261 µg/L to 0.0092 µg/L), 31.8 µg/L for copper (from 49.1 µg/L to 17.3 µg/L), 3.5 µg/L for selenium (from 4.0 µg/L to 0.5 µg/L), and 47.3 µg/L for zinc (from 49.2 µg/L to 1.9 µg/L; Table 4, Figure 10, Figure 11, Figure 12, and Figure 13, respectively). In contrast, molybdenum removal was fairly constant through 2017 (both commissioning-B and operations) with a decrease of 3.9 µg/L (5.7 µg/L to 1.8 µg/L) during commissioning-B and a decrease of 3.6 µg/L (from 6.3 µg/L to 2.7 µg/L) during operations (Table 4). The percent removal of molybdenum and selenium in 2017 is notable, as it has increased from 0% removal during commissioning-A in 2015 and 2016 (Contango, 2016 and 2017) (Table 4). Furthermore, nitrite and nitrate outflow concentrations also decreased from feed water concentrations throughout the demonstration-scale CWTS. Nitrite and nitrate are therefore being removed through treatment in the CWTS (Figure 15 and Figure 16). These results indicate that the commissioning periods were successful in establishing beneficial conditions for the removal of constituents in the CWTS and treatment of these constituents should continue through 2018.

Table 4 – Percent removal of dissolved constituents in the demonstration-scale CWTS.

COC (µg/L)		Commissioning period				Operational period
		2015[1]	2016[2]	2016[3]	2017[4]	2017[5]
Cd	In	0.0505	0.0240	0.0185	0.0163	0.0261
	Out	0.0248	0.0142	0.0066	0.0093	0.0092
	%	**49**	**40**	**64**	**41**	**66**
Cu	In	54.5	61.6	46.1	59.5	49.1
	Out	45.0	56.7	28.8	35.5	17.3
	%	**17**	**8**	**37**	**36**	**65**
Mo	In	11.0	7.6	7.3	5.7	6.3
	Out	11.3	7.6	5.7	1.8	2.7
	%	**0**	**0**	**21**	**62**	**57**
Se	In	11.0	3.7	5.7	2.5	4.0
	Out	11.3	3.4	3.3	0.9	0.5
	%	**0**	**8**	**41**	**61**	**87**
Zn	In	12.3	92.8	37.6	135.3	49.2
	Out	6.8	49.0	11.6	4.9	1.9
	%	**42**	**47**	**69**	**95**	**96**

[1] Values calculated from end of last two sampling event in 2015 before the addition of EDTA (September 9 and 17, 2015; during commissioning-A).
[2] Commissioning-A (May 2, 2016 to July 28, 2016).
[3] Commissioning-B (July 29, 2016 to September 30, 2016).
[4] Commissioning-B (May 27, 2017 to August 17, 2017).
[5] Operational period (August 18, 2017 to September 22, 2017).



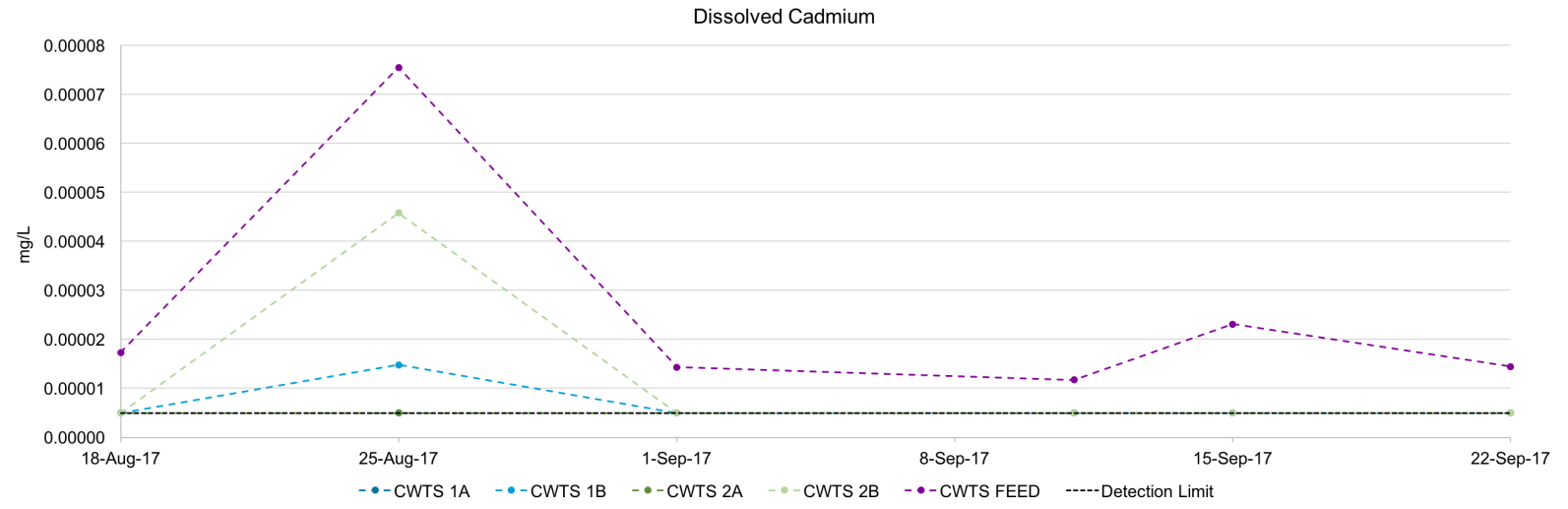
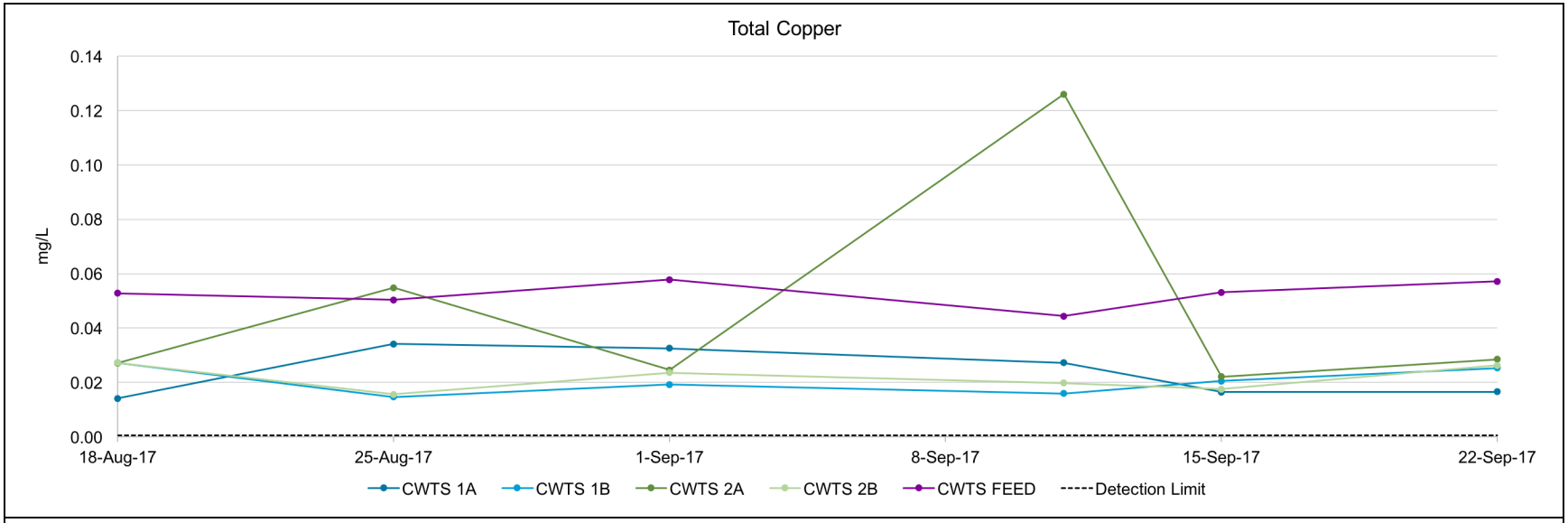


Figure 10 – Cadmium concentrations in operational period of the demonstration-scale CWTS.
The ALS detection limit for cadmium is 0.000005 mg/L.



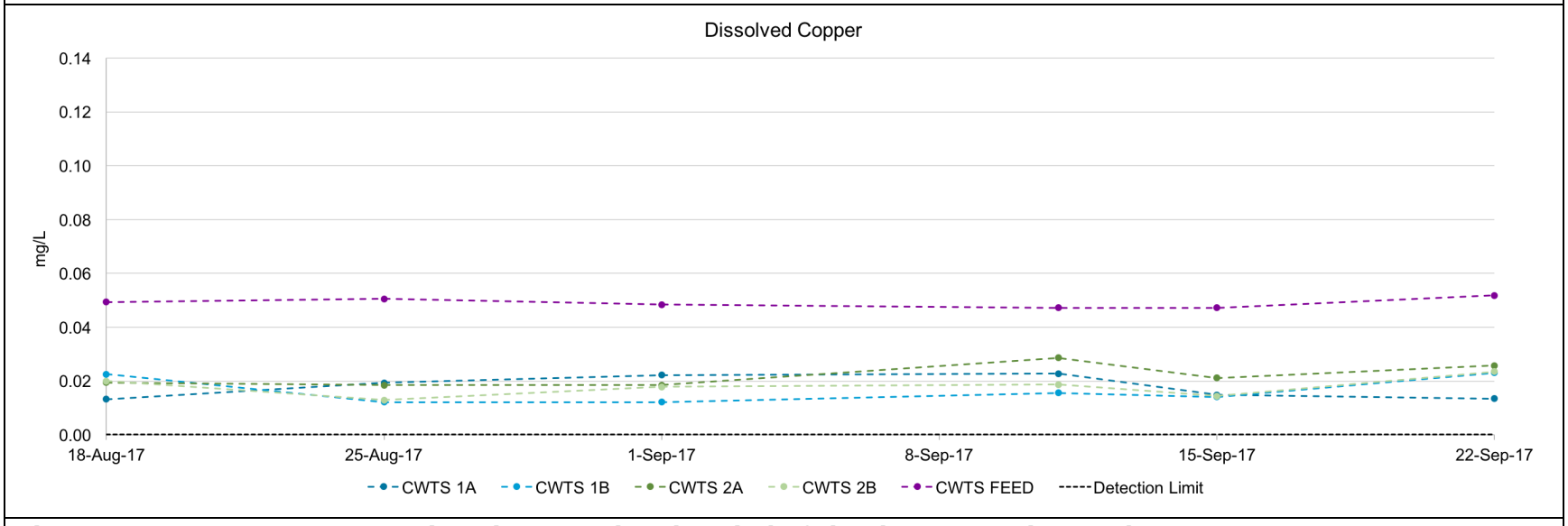


Figure 11 – Copper concentrations in operational period of the demonstration-scale CWTS.
The ALS detection limit for dissolved and total copper is 0.0002 mg/L and 0.0005 mg/L, respectively.

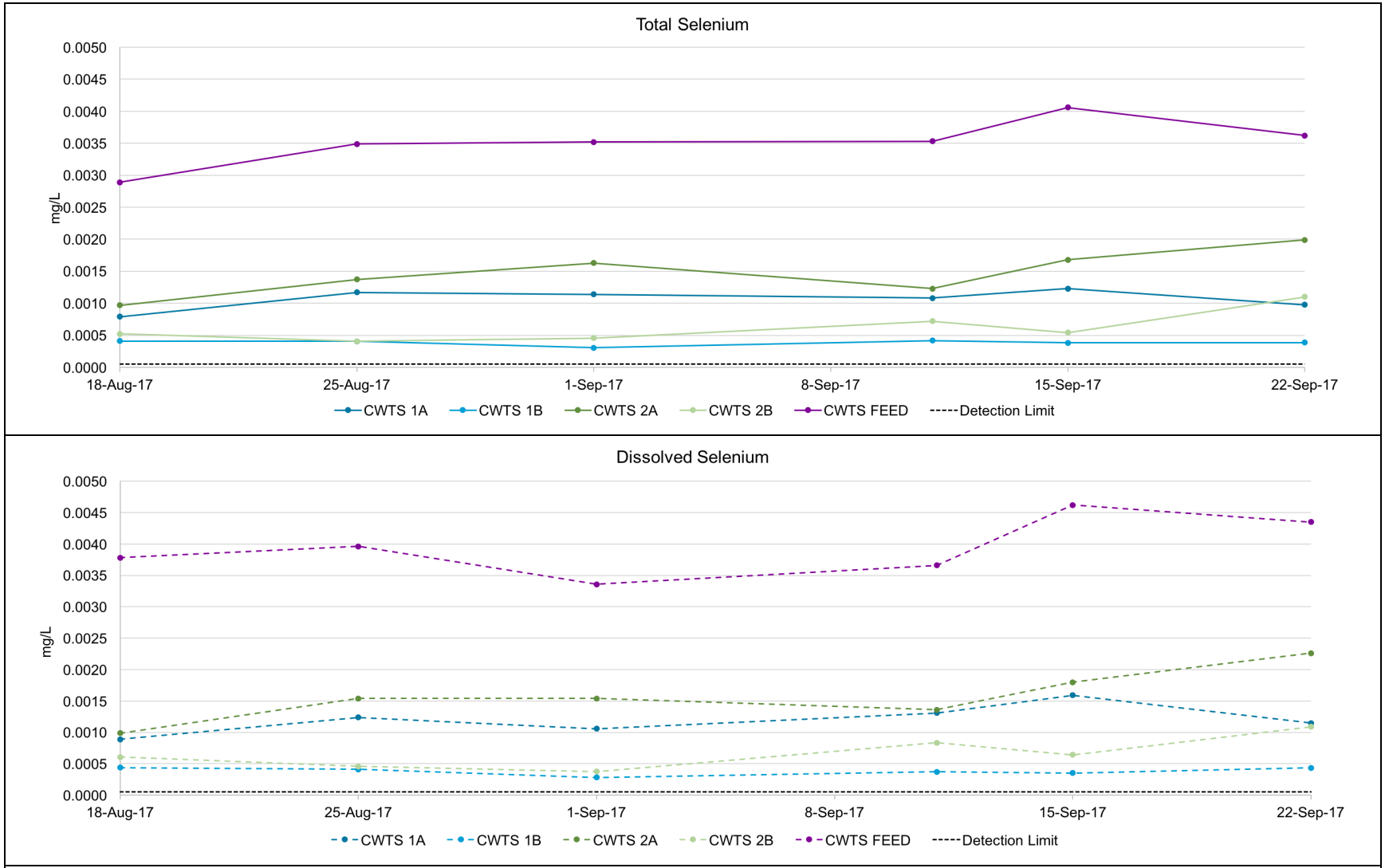
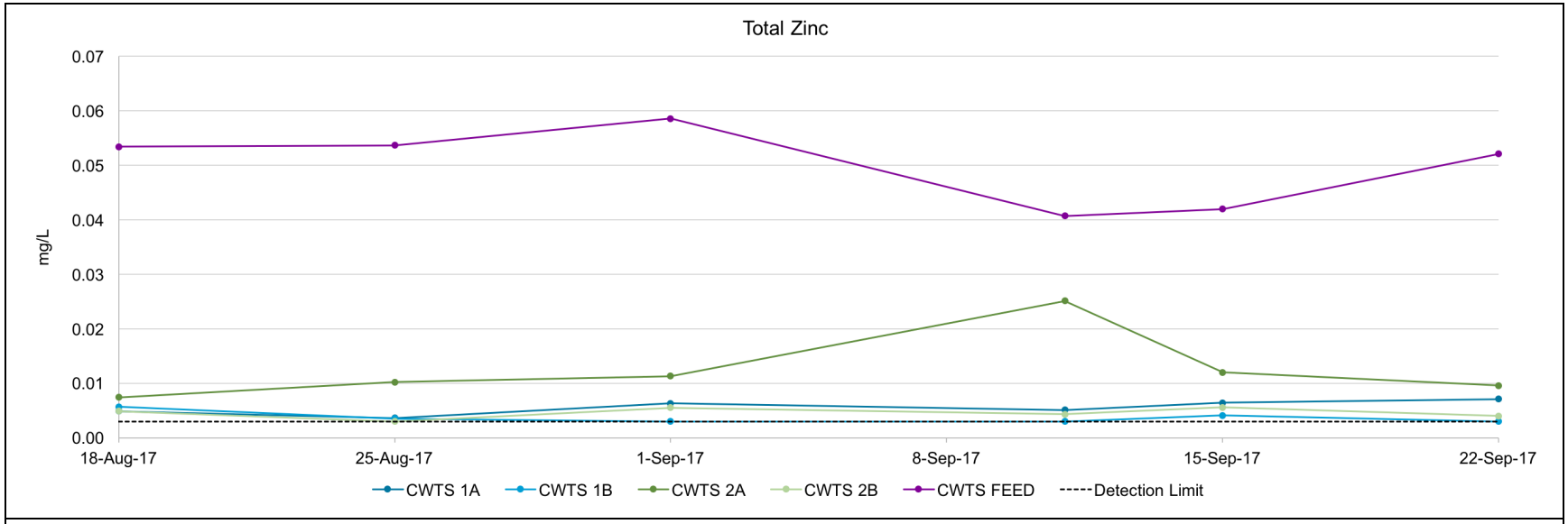


Figure 12 – Selenium concentrations in operational period of the demonstration-scale CWTS.
The ALS detection limit for selenium is 0.000050 mg/L.



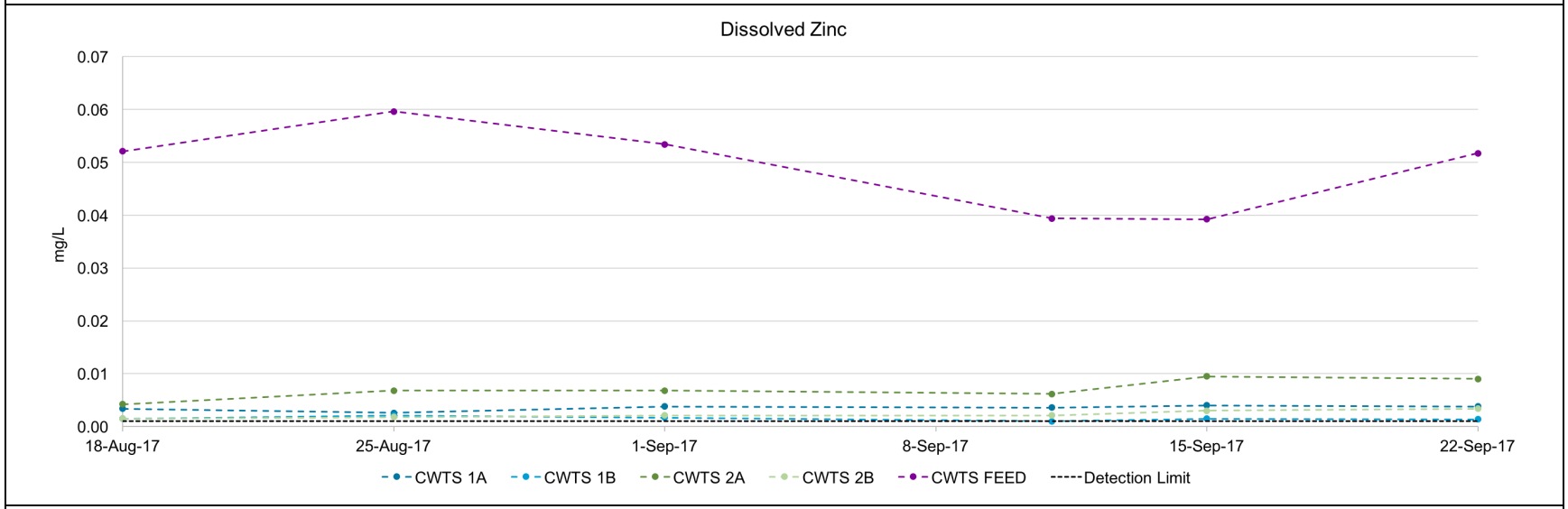
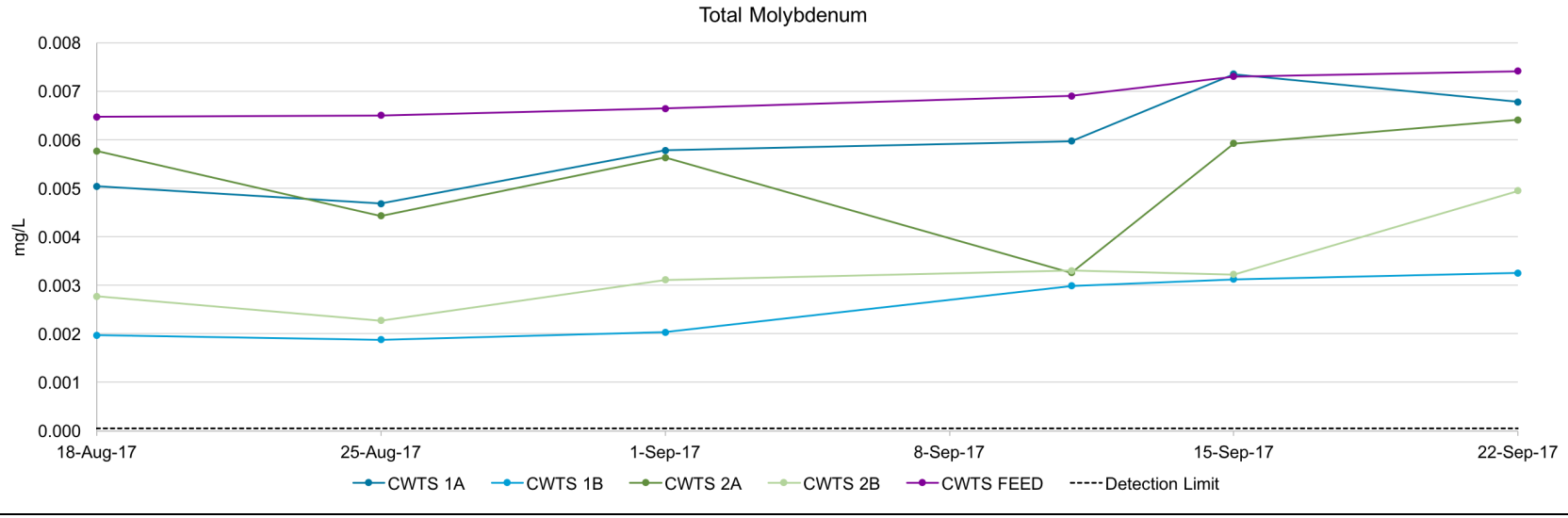


Figure 13 – Zinc concentrations in operational period of the demonstration-scale CWTS.
The ALS detection limit for dissolved and total zinc is 0.001 mg/L and 0.003 mg/L, respectively.



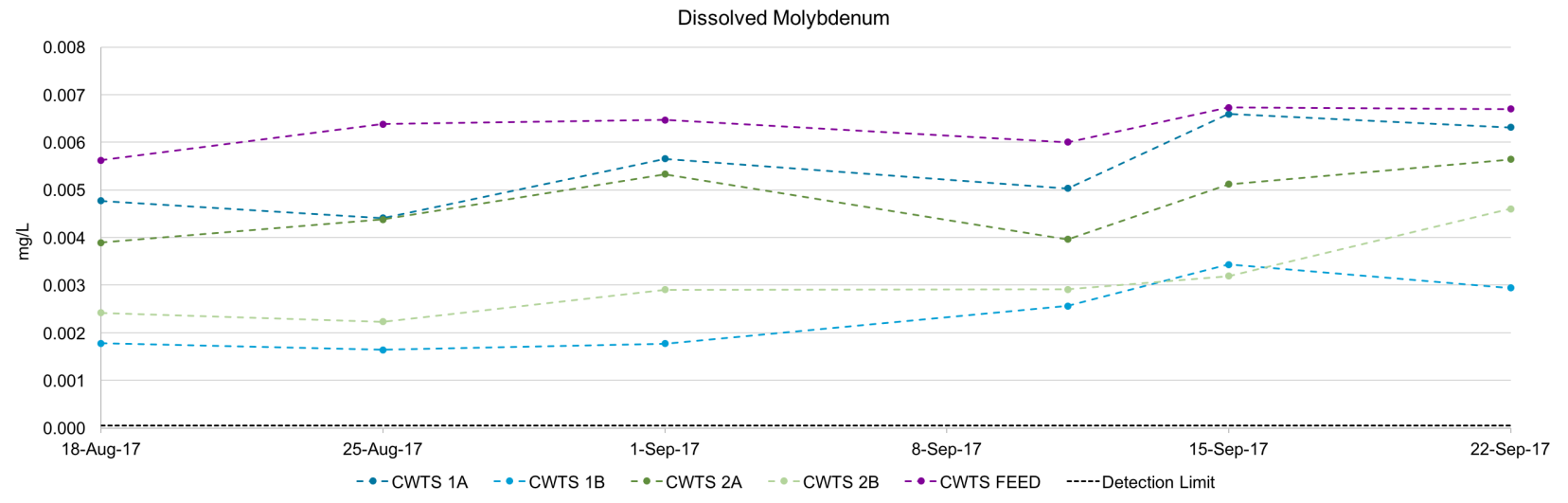


Figure 14 – Molybdenum concentrations in operational period of the demonstration-scale CWTS.
The ALS detection limit for molybdenum is 0.000050 mg/L.

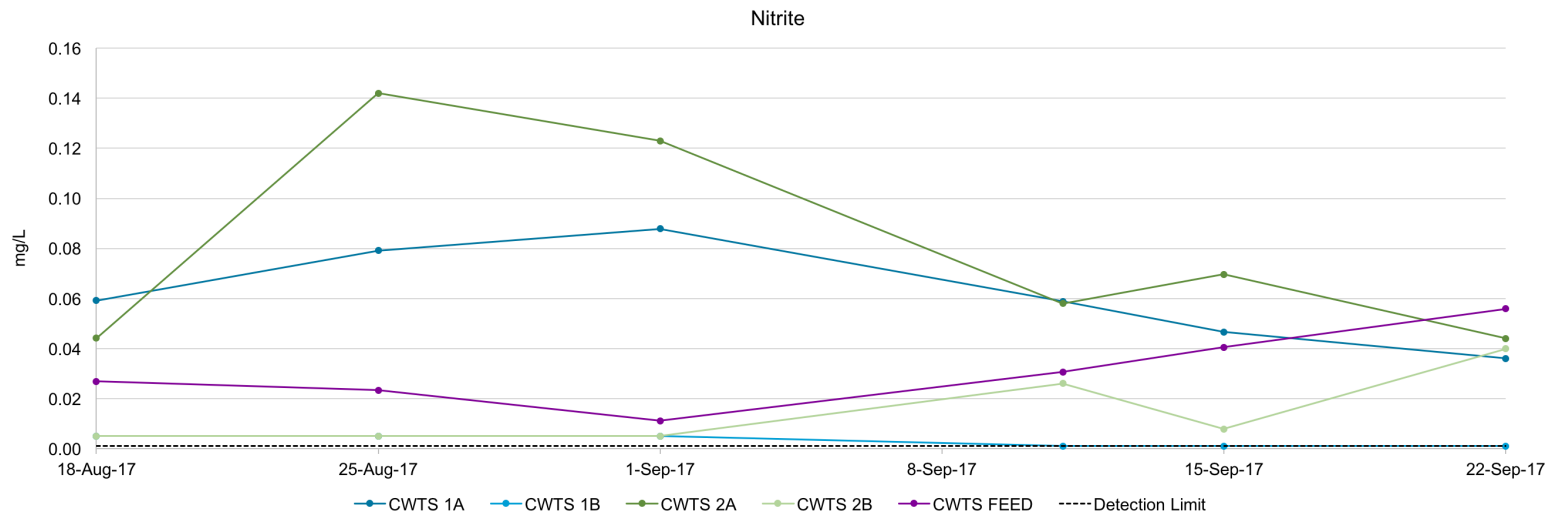


Figure 15 – Nitrite as N (NO_2) concentrations in operational period of the demonstration-scale CWTS.
The ALS detection limit for nitrite is 0.0050 mg/L.

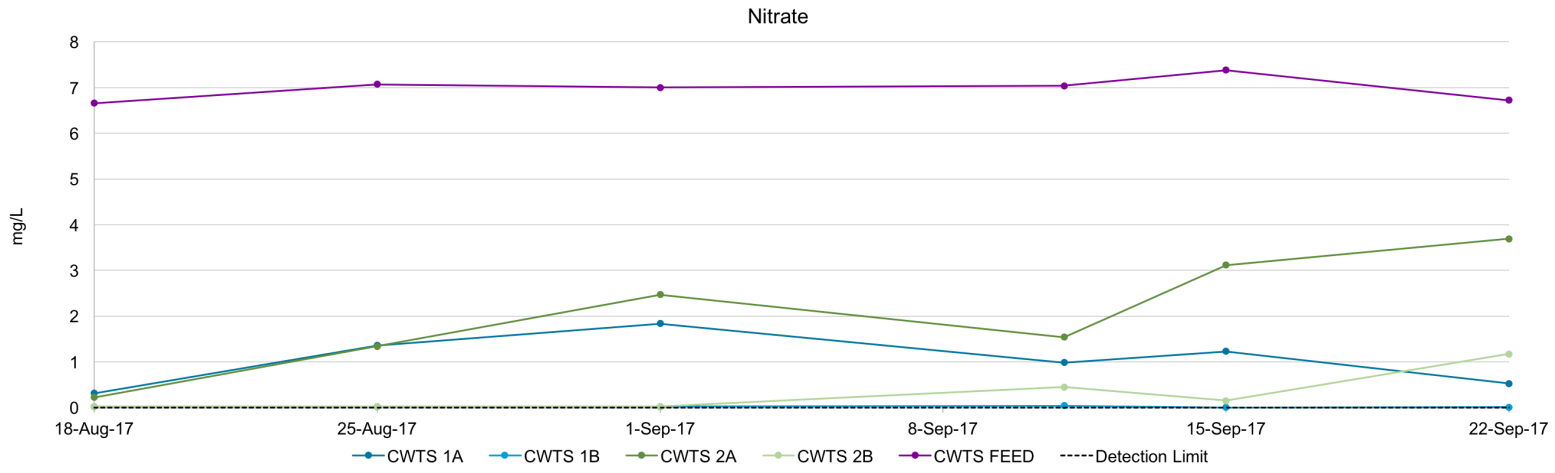


Figure 16 - Nitrate as N (NO_3) concentrations in operational period of the demonstration-scale CWTS.
The ALS detection limit for nitrate is 0.025 mg/L.

5.3. Hydraulic Retention Time

The following are key findings regarding the hydraulic retention time, which are detailed further below:

- The tracer study effectively demonstrated the HRT and flow symmetry through the CWTS.
- There was a single flow path in the CWTS (shown by a single peak in the tracer study)
- Water is incorporating in to the CWTS soils (shown by the long tail for depletion of the tracer).
- The nominal HRT is calculated from the area of the CWTS and the depth at the in situ measuring points. This nominal HRT does not account for depth variations, embankment slopes, vegetation (using space in the water), or substrate pore space involvement. It was found that once all of these factors are in play, the correction factor from nominal to actual is only 0.01 added to the depth of the CWTS, which is incorporated into the HRT calculation as expressed in Equation 3.

5.3.1. Tracer Study

A tracer study was conducted in cell 1A of the CWTS in 2016 (June 13-15, 2016) and in 2017 (June 21-July 2, 2017). Results from the 2016 tracer study can be found in the 2016 demonstration-scale CWTS report (Contango, 2017). The methods of the 2017 tracer study were refined from the 2016 tracer study methods and are further discussed in Appendix A. The nominal HRT is calculated from the area of the CWTS and the depth at the in situ measuring points. This nominal HRT does not account for depth variations, embankment slopes, vegetation (using space in the water), or substrate pore space involvement. The purpose of the tracer studies was to determine the actual hydraulic retention time (HRT) that occurs in the CWTS with a known inflow rate and depth measurement. Then, using this actual HRT form the tracer study, a correction factor can be developed for future CWTS HRT calculations.

Salt (NaCl) was used as the tracer in these tracer studies. By adding a salt solution at the inflow of the CWTS, and placing a YSI meter at the outflow of the CWTS, it could be determined how long it took for the salt to pass through the CWTS by monitoring the specific conductance (SPC) in situ (as salt raises the SPC). The actual HRT of the CWTS was determined by calculating the amount of time that passed between the addition of the salt tracer at the inflow of the CWTS and the peak in SPC at the outflow.

The nominal HRT at the time of the tracer study was calculated based on the flow rate at the time of the tracer study, the known area of cell 1A, and measured depth of cell 1A obtained from the depth sticks that were installed in the CWTS in June 2016 (Appendix A, Figure A5). The results from the 2017 HRT tracer study suggests that the actual HRT of cell 1A is 2.25 days while the nominal HRT of cell 1A is 2.05 days. Therefore, the actual HRT was approximately 10% slower than the nominal HRT. The actual HRT of 2.25 days obtained from the 2017 tracer study was used to solve Equation 1, implying 0.14 m of water involvement

as the "depth", which was 0.01 m greater than the measured depth of 0.13 m of water. This results in a correction factor of 0.01 (Equation 2). Therefore, the correction factor can be applied to the operational HRT as shown in Equation 3 in order to more accurately estimate HRT of the CWTS.

This HRT correction factor has been incorporated into calculations for the operational HRT for the demonstration-scale CWTS in 2017. The correction factor was incorporated into the operational HRT calculations in Section 5.3.2 and used to determine the 2017 removal rate coefficients in Section 5.4.2.

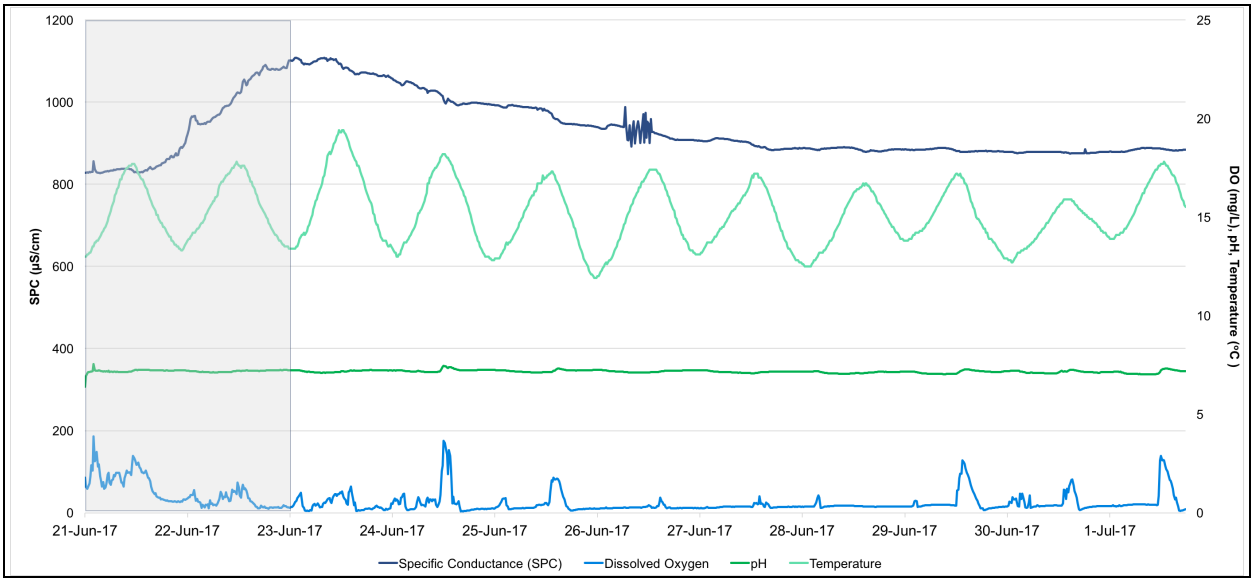


Figure 17 – Hydraulic retention time tracer study results.
Specific Conductance (SPC; µS/cm) on primary y-axis. Dissolved oxygen (mg/L), pH, and temperature (°C) on secondary y-axis. Grey shading indicates time from when CWTS was dosed to the peak of tracer breakthrough, which was used as the actual HRT. The fluctuation in temperature correspond to day (peaks) and night (lows). The dates on the x-axis are at 24 hr intervals, beginning from 7:52 am on June 21, 2017.

$$\text{Actual depth} = \frac{(Q \times \text{actual HRT})}{A}$$

Equation 1 – Equation for calculation of the actual CWTS depth including the correction factor.
Q is the flow rate; actual HRT is the HRT obtained from the tracer study (2.25 days); A is the area of cell 1A.

$$\text{Correction factor} = \text{Actual depth} - \text{measured depth}$$

Equation 2 – Equation for calculation of the correction factor.
The actual depth was calculated via Equation 1 (0.14 m); measured depth is the depth of cell 1A during the tracer study (0.13 m).

5.3.2. Operational Hydraulic Retention Time

The operational HRTs of the demonstration-scale CWTS for 2017 were calculated for each series using Equation 3. Equation 3 uses the volume of a CWTS series including a correction factor (calculated from the results of the tracer study in Equation 2; Section 5.3) and the flow rate from the operational period under consideration. For the operating period in 2017, the average HRT was calculated for August 17 – September 22, 2017 as the flows and depths were relatively stable during this time. The operational HRTs calculated were 5.85 days and 6.82 days for series 1 and series 2, respectively. The operational HRTs were then averaged and used in the calculation of removal rate coefficients in Section 5.4.2.

$$\text{Operational HRT} = \frac{A \times \text{actual depth}}{Q}$$

Equation 3 – Equation for calculation of hydraulic retention time.
Operational HRT is the confirmed hydraulic retention time; *A* is the area of the CWTS; actual depth was calculated using Equation 1 (0.14 m); *Q* is the flow rate.

5.4. Treatment Effectiveness

The following are key findings regarding treatment effectiveness of the CWTS, which are detailed further below:

- All targeted constituents are being treated by mineralization and sequestered to the soils (minimal plant uptake).
- The lowest concentrations consistently achievable for the treatment design (thermodynamic minimums) were reached by the end of the A cells for cadmium and copper.
- RRCs for cadmium and zinc in the 2017 demonstration-scale CWTS were artificially low because low flow rates did not provide the resolution needed to determine a RRC.
- Removal rate coefficients (RRCs, *k*) have been developed that can be used for full-scale sizing.
- Copper leaching from soils has decreased, but is still likely making the RRC artificially low in this CWTS; however, the RRC is expected to improve once copper leaching has subsided.

5.4.1. Thermodynamic Minimums

The thermodynamic minimum is the lowest concentration consistently achievable for a given treatment design and water chemistry. Once reached, making the CWTS bigger will not result in further decrease of outflow concentration (although outflow load may continue to decrease). Therefore, thermodynamic minimums are useful parameters to aid in developing appropriate RRCs, and determining appropriate sizing and outflow water quality objectives of a CWTS. The thermodynamic minimum was determined for each constituent in the demonstration-scale CWTS. The thermodynamic minimum was calculated by comparing the A cell and the B cell concentrations in a series to determine if the concentrations were significantly different or not. If a concentration had stabilized between the A and B cell in a series, the thermodynamic minimum was considered to have been met. To determine when concentrations had stabilized, statistical analyses were done (paired t-test, one-tailed, α=0.05) to determine if the concentrations were significantly different or not. These analyses were then also visually compared to the concentrations of constituents at seven sampling locations throughout the CWTS from two time points, to further evaluate and confirm whether concentrations appeared to have leveled off (Figure 8 and Figure 9 for copper and selenium, respectively; Figures B1 to B4 in Appendix B for remaining constituents). However, there was not enough time points in the operational period for this data to be used in the statistical analyses to further refine the expected HRT for the thermodynamic minimums. This can be refined in 2018, by varying flow rates.

Based on the results of the t-test for the A cell and B cell concentrations, it was determined that the thermodynamic minimum was achieved for cadmium and copper by the end of the A cells in the demonstration-scale CWTS. The detection limit was reached for cadmium prior to the outflow of cell A, suggesting that faster flows would be needed to accurately determine the HRT at which the thermodynamic minimum was met (Figure 18). The thermodynamic minimum for molybdenum, nitrate, and selenium were not yet statistically achieved,

suggesting that a larger CWTS would achieve lower concentrations of these constituents (Appendix B, Figures B13 to B17). The thermodynamic minimum for zinc was not statistically achieved, however, the majority of zinc load was removed within the beginning of cell A, with minimal load removal in the remainder of the CWTS (Figure 18). Furthermore, outflow concentrations of both cell A and B for zinc are at or near detection limit (Figure B17), so it is not necessarily beneficial to build a larger CWTS for zinc.

As shown by the graphs previously mentioned, concentrations in A cells and B cells were very similar for cadmium and copper. This shows that treatment is occurring in the A cells and has reached a thermodynamic minimum which does not decrease further for the B cells. However, these results for copper should be taken with consideration that there is still some residual copper in the soils from construction that is being treated, and therefore, there is more copper treatment ongoing that can be noted by just the water concentrations.

Interestingly, once cadmium, copper and nitrate are treated, zinc and molybdenum begin to have an increase in treatment. This phenomenon is referred to as a treatment front, which generally abides by reaction orders (i.e., nitrate treatment before sulphate, etc.; Murray-Gulde, 2008).

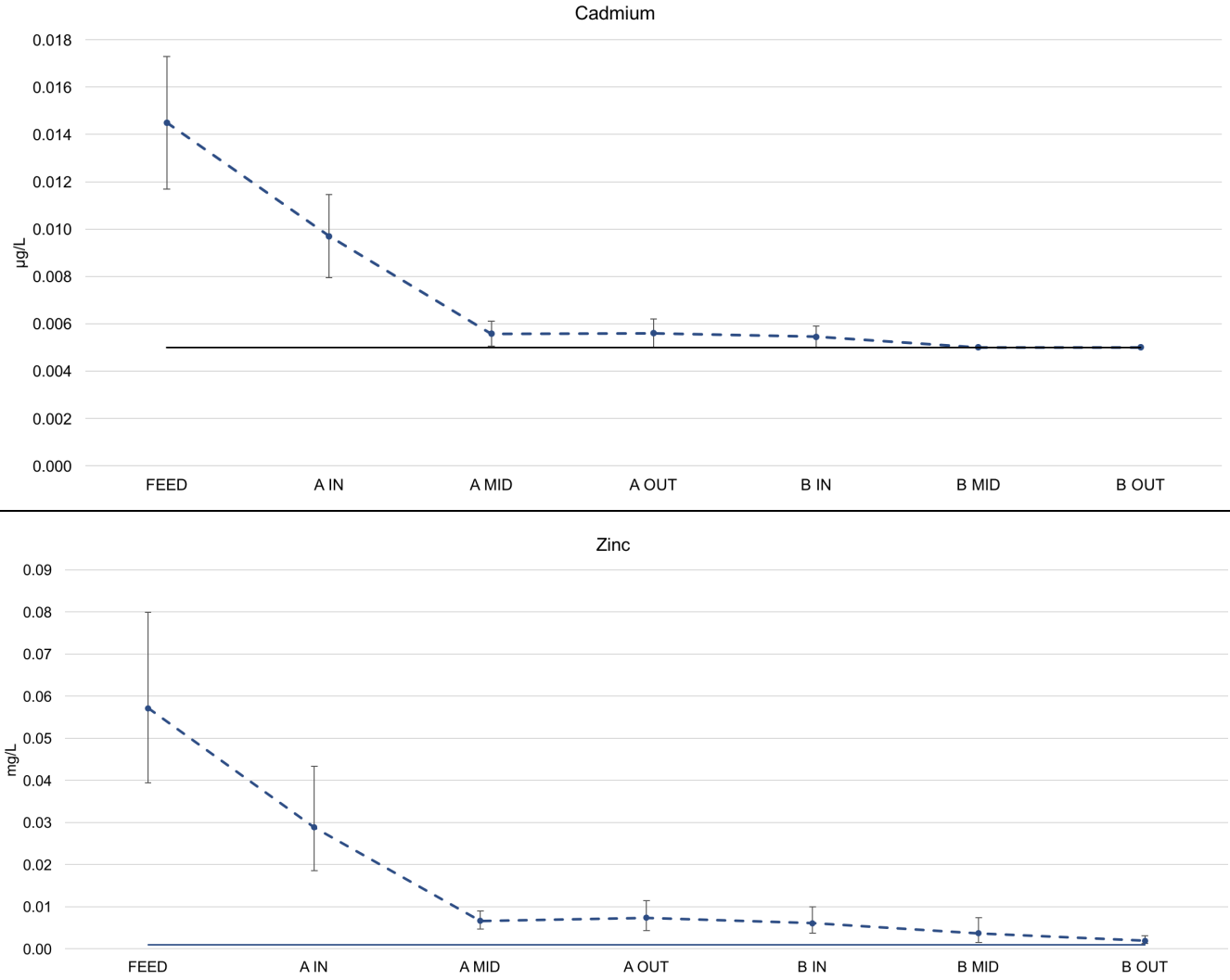


Figure 18 – Average dissolved cadmium and zinc concentrations through the CWTS during the operational period. Data shows the average of series 1 and series 2 for two sampling points during the operational period (August 18 and September 11, 2017). Error bars show minimum and maximum concentrations at each sampling point. The black horizontal line shows the detection limit of 0.005 µg/L for cadmium and 0.001 mg/L for zinc.

5.4.2. Removal Rate Coefficients

An important factor for CWTS design is the rate of treatment, also known as the removal rate coefficient (RRC; *k*). The RRC is based on the treatability of a specific compound and the hydraulic retention time of the CWTS, both of which are site-specific based on water chemistry, CWTS designs, and characteristics of the CWTS. A RRC was calculated and applied for cadmium, copper, molybdenum, selenium, zinc, and nitrate which can be used to refine future sizing and performance estimations of a full-scale CWTS (Table 5). In pilot-scale testing specific for the Minto CWTS, the RRC for selenium was a zero-order reaction kinetic, however, optimizations of the operation of the system through the demonstration-scale commissioning have enabled a first-order RRC to be maintained. Therefore, cadmium, copper, selenium, zinc, and nitrate are calculated as first-order kinetics, but molybdenum followed a zero-order kinetic. In other words, the reaction rate for molybdenum is a constant rate and does not depend on concentration, whereas the reaction rates for cadmium, copper, selenium, zinc, and nitrate are proportional to concentration (a half-life type of reaction).

In Equation 4 and Equation 5, *Cf* is the final concentration, *Ci* is the initial concentration, and *HRT* is the operational hydraulic retention time calculated in Equation 3 (Section 5.3.2) using the CWTS depth and correction factor calculated from the tracer study (Equation 1 and Equation 2; Section 5.3).

$$k = \frac{-\ln\left(\frac{Cf}{Ci}\right)}{HRT}$$

Equation 4 – Equation for calculation of first-order removal rate coefficient.
k is the removal rate coefficient; *Cf* is the final concentration; *Ci* is the initial concentration; *HRT* is the operational hydraulic retention time.

$$k = \frac{(Ci - Cf)}{HRT}$$

Equation 5 – Equation for calculation of zero-order removal rate coefficient.
k is the removal rate coefficient; *Cf* is the final concentration; *Ci* is the initial concentration; *HRT* is the operational hydraulic retention time.

RRCs were calculated for each constituent (dissolved and total) and series 1 and series 2 were averaged for the operational period in 2017 and compared to RRCs calculated to the commissioning period of 2016, and pilot-scale RRCs (Table 6). In 2016, all RRCs were calculated using outflow concentrations from the B cells while in 2017, cadmium and copper RRCs were calculated using outflow concentrations and HRTs from the A cells as they had reached their thermodynamic minimum at the end of cells 1A and 2A (Section 5.4.1). The RRCs for the remaining constituents did not reach their thermodynamic minimum in the A cells and, therefore, outflow concentrations and HRTs from the B cells were used. RRCs for cadmium and zinc in the 2017 demonstration-scale CWTS were artificially low because low flow rates did not provide the resolution needed to determine a RRC. Additionally, RRCs for copper in demonstration-scale CWTS are also artificially low due to copper leaching from soils used in construction of the system. Therefore, the RRCs for copper are expected to improve

once leaching has subsided. In general, RRCs improved from those calculated in 2016 (Table 5; Contango, 2017).

Applying the removal rate coefficients (k) from Table 5 to Equation 3 above, parameters can be rearranged to solve for those of interest, such as the volume needed, that in turn determines the area of CWTS required which is dependent upon the design. Analytical results from August 17 – September 22, 2017 were chosen to calculate k because this is the operational period after commissioning-B was completed and flow rates were stable and at targeted values. The treatment rate coefficients applied here are intended to be conservative estimates for conceptual sizing purposes, and will need to be refined through further demonstration-scale (on site) testing.

Table 5 – Removal rate coefficients (*k*) for constituents of concern.

COC	Pilot[1] *k**day^{-1}	2016 Demonstration[2] (Commissioning) *k**day^{-1}		2017 Demonstration[3] (Operational) *k**day^{-1}		Recommended[4] *k**day^{-1}
	Total	Dissolved	Total	Dissolved	Total	Dissolved
First order removal rate coefficients						
Cd	1.142	0.172	0.174	>0.521[5]	>0.226[5]	1.142
Cu[6]	1.171	0.0778[6]	0.0740[6]	>0.3100[6]	>0.1356[6]	1.171
Mo	0.1010	0.0416	0.0468	N/A	N/A	N/A
Se	N/A	N/A	N/A	0.3186	0.3069	0.3186
Zn	1.144	0.195	0.183	0.512[5]	0.394[5]	1.144
Nitrate	0.559	0.0914	N/A	0.591	N/A	0.591
Zero order removal rate coefficients						
Mo	N/A	N/A	N/A	0.0006	0.0006	0.0006
Se	0.00181	0.000357	0.000316	N/A	N/A	N/A

[1] Values calculated from data in Contango, 2014b (pilot-scale report) for *Carex aquatilis* + moss with low nitrogen scenario.
[2] Analytical results from August 10 – September 17, 2016 were used.
[3] Analytical results from August 18 – September 22, 2017 were used.
[4] Cadmium, copper, and zinc recommended RRCs are developed from total concentrations in pilot-scale testing and are conservative proxies for dissolved concentrations.
[5] RRCs for cadmium and zinc in the 2017 demonstration-scale CWTS were artificially low because low flow rates did not provide the resolution needed to determine a RRC, so the lowest potential *k* value based on available data is reported here.
[6] RRC's for copper in demonstration-scale CWTS are artificially lowered due to copper leaching from soils used in construction of the system and are therefore expected to improve once leaching has subsided.
N/A = an RRC is not available. All RRCs are for first-order reaction kinetics except for selenium in the 2016 demonstration-scale CWTS and molybdenum in the 2017 demonstration-scale CWTS, which are zero-order reaction rate kinetics. All RRCs in 2016 and 2017 used outflow concentrations for the B cells in the CWTS.

5.5.Evapotranspiration in the Demonstration-Scale CWTS

The following are key findings regarding the evapotranspiration studies, which are detailed further below:

- The evapotranspiration studies revealed a significant loss of water, which will impact calculations of loads to the receiving environment (making them lower than previously estimated).
- In May and June, an average water loss of 5.3 L/day/m^2 was observed, which is equivalent to 18-20% of water (~700 L/day lost in the demonstration CWTS).
- During the evapotranspiration trials, copper leached into the water as it was transformed from an oxide mineral to a sulphide mineral (because of the copper in the soils used for construction). This is not representative of what would occur during periods with no flow in a full-scale CWTS, where soils with minimal leachable copper are used and copper is deposited in sulphide form (fraction 4) by the biogeochemical activity of the CWTS (Section 5.6).

Total evapotranspiration from a CWTS is measured as the combined effects of open water evaporation and plant transpiration (Beebe et al., 2014). The purpose of calculating the evapotranspiration of a CWTS is to understand the amount of water lost per day over the surface area of a wetland, which in turn concentrates constituents, and should be considered in the context of the decrease in outflow concentration (or not) and outflow load reduction. It should also be noted that evapotranspiration will vary depending on temperature, humidity, and wind on-site and, therefore, will fluctuate over time. This should be considered when interpreting evapotranspiration rates, however, for conceptual calculation purposes, a set rate is applied here.

The off-site evapotranspiration study conducted in 2016 suggested that approximately 15.5 L/day/m^2 could be lost due to evapotranspiration (Contango, 2017). The results of the 2017 on-site evapotranspiration studies were intended to provide more accurate estimates at the CWTS than the off-site study and had an average water loss of 5.3 L/day/m^2 (Table 6). These results are consistent with pilot-scale results from another CWTS planted with *Carex* where 5.1 L/day/m^2 were lost through evapotranspiration. Although the off-site study can be informative, the on-site study is more representative of what would occur during operation of the full-scale CWTS in the spring. The difference in the two rates is expected to be due to temperature difference of the trial. The on-site during the evapotranspiration study were on average 9.6°C in May and 13.7°C in June while temperatures in the off-site study were on average 22°C during the day (12 hr) and 16°C during the night (12 hr) which is less representative of site conditions. Temperatures in July and August at Minto were only slightly higher than June temperatures and were on average 15.8°C and 13.9°C, respectively. Therefore, we would expect marginally higher rates of evapotranspiration in these months.

Table 7 shows the amount of water lost to evapotranspiration during the operational period (August 18 – September 22, 2017) in each cell of the demonstration-scale CWTS. The amount of water lost in the CWTS during each flow rate period tested is calculated using Equation 6.

$$Water\ loss\ (\%) = \frac{ET}{Q} \times 100$$

Equation 6 – Equation for the calculation of the water loss due to evapotranspiration.
Q is the flow rate (L/day); *ET* is the evapotranspiration rate (L/day).

Table 6– Water loss through evapotranspiration in the CWTS.

Study	Cell	Water loss[1] (L/day)	Water loss[2] (L/day/m^2)
Trial 1: May 16 – 24, 2017	1A	575.0	8.3
	1B	143.7	2.1
	2A	509.0	8.8
	2B	152.7	2.6
		L/day/m^2 Average	**5.4**
Trial 2: June 8 – 11, 2017	1A	580.8	8.3
	1B	116.2	1.7
	2A	775.7	13.3
	2B	339.4	5.8
		L/day/m^2 Average	**7.3**
Trial 3: June 11 – 15, 2017	1A	292.7	4.2
	1B	27.9[3]	0.4
	2A	273.4	4.7
	2B	203.6	3.5
		L/day/m^2 Average	**3.2**
		L/day/m^2 Overall Average	**5.3**

[1] Water loss (L/day) = (initial depth (m) – final depth (m)) * area of CWTS (m^2) * 1000 L/m^3/days.
[2] Water loss (L/day/m^2) = Water loss (L/day) / area of CWTS (m^2).
Total days of study in trials 1, 2, and 3 were 8, 3, and 5 days, respectively. The area of series 1 is 69.7 m^2 and the area of series 2 is 58.2 m^2.
[3] Depths in this cell had minimal change throughout trial 3.

Table 7 shows how this water loss likely affected the load leaving the wetland. The average ambient temperature on-site for this time period was 9.9°C, and so the evapotranspiration rates from the trials are expected to be relevant to the operational period. The information in Table 7 is calculated based on Equation 7 and Equation 8.

Evapotranspiration has a significant effect on the calculation of constituent load to the receiving environment. Therefore, future models for assimilative capacity in the downstream receiving environment should take into account not only the predicted outflow concentrations from the CWTS using removal rate coefficients (Section 5.4.2), but also adjust the load accounting for evapotranspiration.

Concentrations of copper (and other metals) were measured in the water during the evapotranspiration studies to determine if any leaching occurred during periods of no flow because the soils used in construction contained oxidized copper minerals. In the 2017 study, this is noted by an average decrease of 63.3 µg/L copper after flow was restarted and treatment of copper resumed. This is not representative of what would occur during periods

with no flow in a full-scale CWTS, where soils with minimal leachable copper are used and copper is deposited in sulphide form (fraction 4) by the biogeochemical activity of the CWTS (Section 5.6). In a full-scale CWTS, lower flows and stagnation will result in greater treatment of constituents into sulphide mineral forms.

The demonstration-scale evapotranspiration studies were conducted to inform evapotranspiration rate estimates for full-scale CWTS design. However, these results are fortuitous because the study was conducted during the commissioning-B period, rather than the operational period when the CWTS was running optimally. Additionally, since the studies were conducted in the spring, they do not capture the seasonality of evapotranspiration rates and are therefore conservative numbers. Thus, it is recommended that another evapotranspiration study be conducted on-site in 2018.

$$Load_{in} = C_i \times V$$

Equation 7 – Equation for the calculation of load of a constituent into a CWTS over a period of time.

$Load_{in}$ is the mass of a constituent that enters a CWTS over a period of time; C_i is the inflow concentration of the constituent; V is the volume of water that enters a CWTS over a period of time.

$$Load_{out(ET)} = Water\ Out_{ET} \times C_f$$

Equation 8 – Equation for the calculation of the load out of a constituent over a period of time, adjusted for evapotranspiration.

$Load_{out(ET)}$ is load out of a constituent over a period of time, adjusted for evapotranspiration; $Water\ Out_{ET}$ is the outflow water volume over a period of time, adjusted for evapotranspiration; C_f is the outflow concentration of the constituent.

Table 7 – Average constituent load in, load out, load out adjusted for evapotranspiration and load removed adjusted for evapotranspiration in the demonstration-scale CWTS.

COC	Form	Load in (mg/ day)	Load out[1] (mg/day)	Load out Adjusted for evapotranspiration[2] (mg/day)	Load removed adjusted for evapotranspiration (mg/day)	Load removed[3] (%)
Cd	D	0.0476	0.0162	0.0131	0.0345	71
	T	0.0283	0.0094	0.0076	0.0207	73
Cu	D	89.7	31.4	25.4	64.3	72
	T	96.2	38.4	31.1	65.1	67
Mo	D	11.5	4.8	3.9	7.6	65
	T	12.6	5.2	4.2	8.3	66
Se	D	7.2	0.9	0.7	6.5	89
	T	6.4	0.9	0.7	5.7	88
Zn	D	90.0	3.4	2.8	87.2	97
	T	91.6	7.5	6.0	85.5	93

[1] Inflow volume was used for the calculation of load out using outflow concentrations without adjusting for evapotranspiration (Appendix A, Equation A2).
[2] Outflow volume adjusted for evapotranspiration was used for the calculation of load out adjusted for evapotranspiration.
[3] Load Removed (%) = (Load out adjusted for evapotranspiration)/(Load in) * 100
Water loss for each flow period is an average of all time points and is calculated using Equation 7. Average water loss in series 1 and 2 is 377.9 L/day/m2 (18%) and 315.4 L/day/m2 (20%), respectively. D = dissolved, T = total. Results in the table are from the operational period only (August 18 – September 22, 2017) and are therefore not representative of what would occur year-round. Ambient temperature on-site for this period was on average 9.9°C.

5.6.Soils

The following are key findings regarding constituents in the soil, which are detailed further below:

- In 2017, leachable copper concentrations in soils decreased in the top 0-10cm while total copper concentrations increased.
- Most constituents, including copper, have shifted primarily into stable reduced and residual minerals fractions in the soil.
- Acid volatile sulphides (AVS) were non-detectable in the CWTS in 2016. In 2017 small amounts of AVS were detected in cells 1A and 2A which indicates that residual sulphides are starting to become available for metal treatment and that copper in the soils are becoming rendered inert through sulphide mineralization.

Although unintentional, the high initial leachable copper concentrations in the CWTS soils (Appendix A, Table A2) allowed for additional testing to be carried out on these CWTSs. Because the soil substrates used for construction of the CWTS were from overburden sources, the copper was in oxidized form rather than in a mineral form that would typically be found in a reducing CWTS (i.e., soils with negative redox). Therefore, there was some initial leaching of copper from the soils into the water.

To assess the effect of the elevated metals in the soils used in construction on CWTS functionality, four soil analytical test methods were used and are described in Appendix A. Analytical methods tested for total metals, leachable metals, and metal speciation. Results of these test methods during the operational period are discussed in the following sections. Table 8 shows the definitions of extractable fractions from the sequential extraction procedure for the speciation of metals (Tessier et al., 1979).

Table 8 - Summary of extractable fractions from sequential extraction procedure for the speciation of metals[1].

Fraction	Description	When COCs Become Unstable and Release to the Water Column
1	Exchangeable fraction for adsorbed minerals	Readily released (i.e., soluble and exchangeable)
2	Mineral fraction bound to carbonates or solubilized at pH 5	Decreased pH
3	Oxidized mineral fraction bound to Fe-Mn oxides	Reducing conditions
4	Reduced mineral fraction and sulphides	Oxidizing conditions
5	Residual mineral fraction (primary and secondary minerals)	Not expected to be released in solution over time under conditions normally encountered in nature

[1]Method based on Tessier et al., 1979.

Analysis of leachable metals in 2017 used the same leach method as in 2016 and results can be found in Appendix B, Table B2. When compared to early 2016 results, leachable copper concentrations in 2017 soils decreased throughout the CWTS (Figure 19) while total copper concentrations increased overall as copper shifted from its leachable to mineral form (Figure 20). In general, soils in the demonstration-scale CWTS appear to be reaching a steady-state with concentrations of leachable copper changing little throughout 2017. We expect leachable copper decrease further in 2018.

Sequential extraction procedure for the speciation of particulate trace metals was conducted for all constituents of concern and results and figures can be found in Appendix B (Tessier et al., 1979). The analysis shows that despite elevated initial leachable copper concentrations, the soils have become more stable (less leachable) over time in the CWTS setting as the soils have aged, shifting from oxidized to reduced, mineral forms (Figure 21; Figures B18 and B19). This beneficial aging of soils to a less soluble mineralized form of sulphide is expected for this type of treatment CWTS design. It should also be noted that copper leached from the original substrate into the water put additional treatment demands on the CWTSs.

For the CWTS to treat the copper from the soils, additional organic carbon is needed beyond that necessary for the waters alone. Organic carbon is contributed to the CWTS through decomposition of plant material as the CWTS matures. Organic carbon is used as food by the microbes, which in turn produce sulphides which drive metals treatment through mineralization that is helped by biogeochemical processes. Presently, as there is excess copper in the soils in oxidized forms transitioning to reduced forms, these will bind to and consume the sulphides produced, transforming them to the sulphide-bound fraction 4 (Table 7). This in turn impacts the ability to treat water, as it uses the available sulphides that would otherwise treat metals in the water. When the mineral fractions of the soils have completed their transformation into sulphide-bound minerals, the soils will no longer be "sulphide hungry", allowing for the soil-produced sulphides to be utilized for metals treatment in the water. This progress can be tracked in results from the sequential extraction tests, the appearance of acid volatile sulphides (AVS) in the soils (iron sulphates, which will only occur after the copper is transformed), and the change of sulphate concentrations in the water.

Additional reserve treatment capacity can be stored in a CWTS through creation of a reserve of AVS (newly formed amorphous iron sulphides). AVS are referred to as such because of the testing method, which is also done alongside simultaneously extracted metals (AVS:SEM; at Minto, cadmium, copper, and zinc were tested for) to determine the ratio of iron sulphides to free metals. An excess of AVS suggests the metals would be non-bioavailable and not likely to leach. In the case of Minto, where the soils used in construction had excess copper, the appearance of measurable AVS would also indicate that the copper in the soils is nearing an endpoint of transformation to sulphide forms.

AVS was tested for in 2016 prior to adding the straw and wood chips and was non-detectable (hence, the decision to add the straw and wood chips). The appearance of small amounts of measurable AVS over time (cell 1A in June and cell 2A in September, 2017) indicates that the amount of sulphide produced in the soils is beginning to exceed the total amount of copper

placed with the original soils and the CWTS should eventually start performing the way it should have if substrates with high leachable copper had not been used. However, the AVS:SEM ratio is indicating there is significant copper in the soils still consuming the sulphides, and treatment performance is expected to improve as that continues to be remedied over time. An AVS:SEM ratio greater than one indicates excess copper in the soils has been reduced to sulphide form and is no longer consuming excess sulphides, allowing for AVS to form. Once the AVS:SEM ratio of all cells is consistently greater than one, we do not expect any further copper leaching to occur, and copper treatment within the CWTS should improve until this time.

Zinc (Figures B20 and B21) mineral forms have become stable over time, with most of these constituents found in the reduced or mineral form (fraction 4 and 5) by the end of 2017. Cadmium concentrations are at or near detection limits for all fractions (Figures B22 and B23). Molybdenum was mostly found in the most stable residual mineral fraction 5, which contained around 40% of the total molybdenum in the samples (Figures B24 and B25). Selenium was at or near the detection limit for all fractions and at the end of 2017, and was barely detectable for the stable fractions 4 and 5 (Figures B26 and B27).

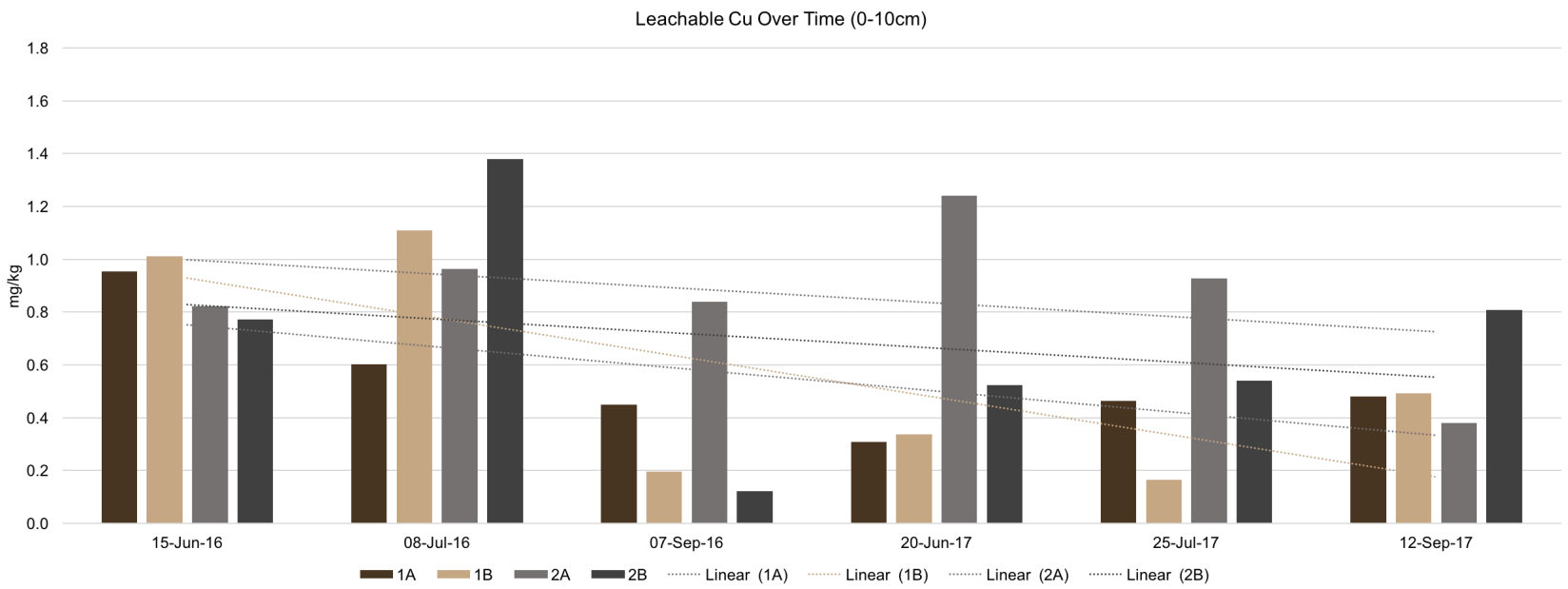


Figure 19 – Leachable copper concentrations in soils over time in the demonstration-scale CWTS.
Figure shows concentrations in shallow soil (0-10 cm). Trendlines (dotted lines) show a general decrease in leachable copper over time.

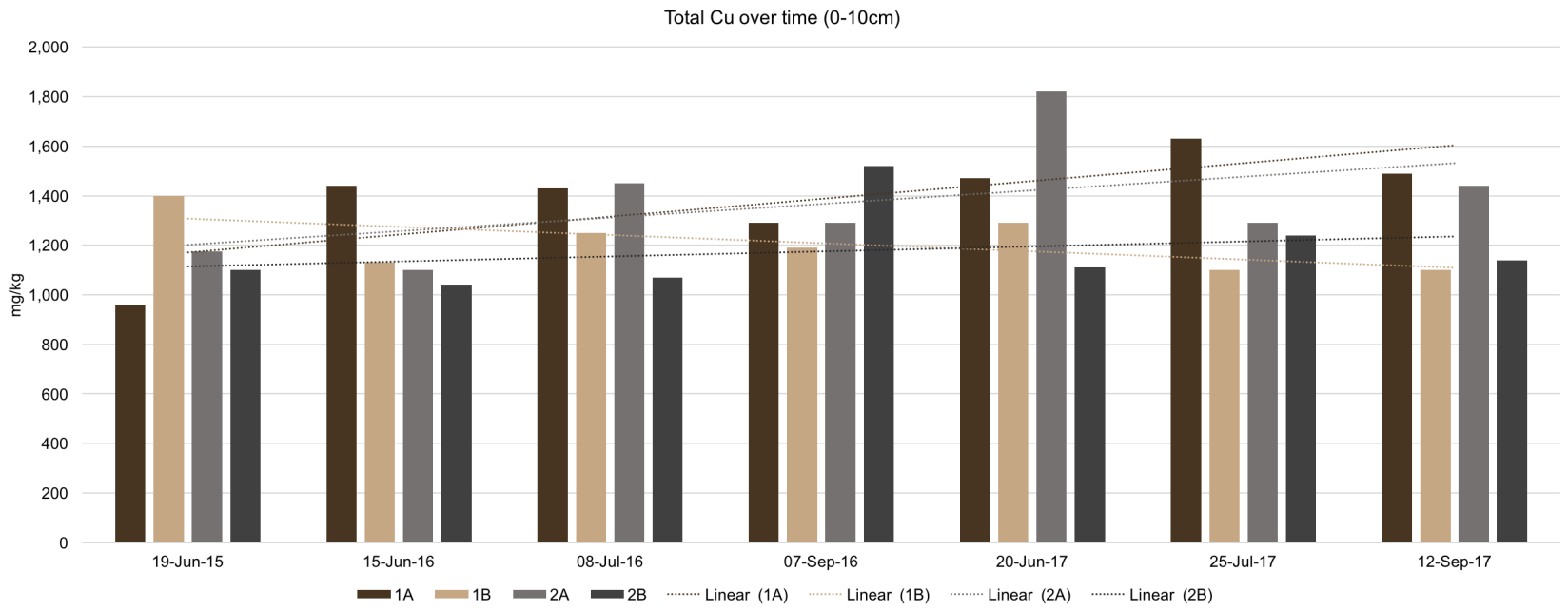


Figure 20 – Total copper concentrations in soils over time in the demonstration-scale CWTS.
Figure shows concentrations in shallow soil (0-10 cm). Trendlines (dotted lines) show a general increase in total copper over time.

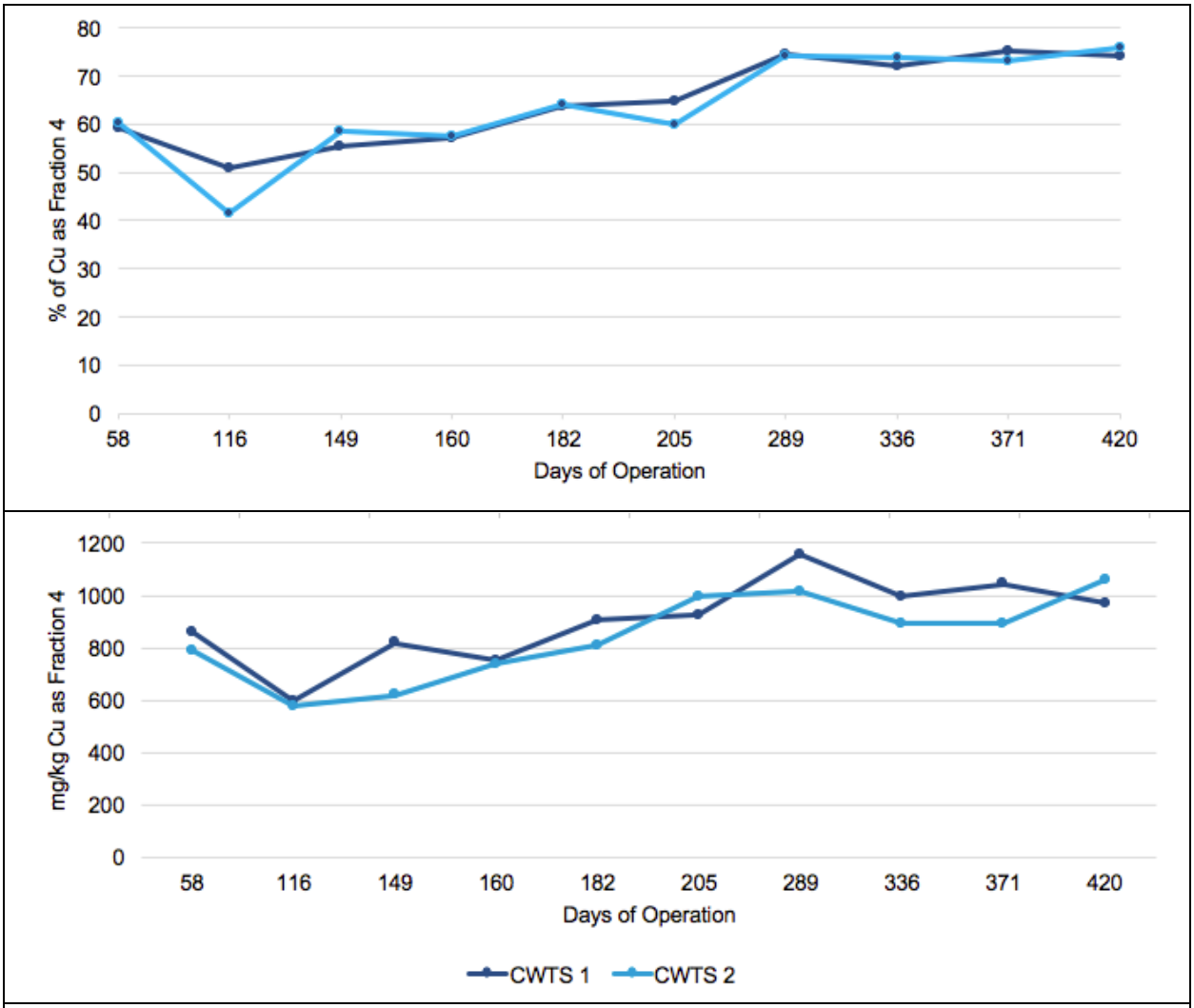


Figure 21 – Copper as sequential leach fraction 4 over time.
Copper in the form of the targeted Fraction 4 (sulphide mineral form and bound to organics), increased in the CWTS over time as the soils matured. Additional information can be found in Appendix B.

5.7.Metals Uptake

The following are key findings regarding metals uptake, which are detailed further below:

- Overall, uptake of metals in *C. aquatilis* was low and generally lower in 2017 than in 2016 demonstration-scale CWTS results.
- Targeted treatment mechanisms are becoming more robust, rendering elements non-bioavailable.

Concentrations of the constituents of concern in *C. aquatilis* and moss from the demonstration-scale CWTS in 2017 were compared to those from 2016. *C. aquatilis* and moss in the demonstration-scale CWTS had greater copper concentrations than the pilot CWTSs, likely reflecting the bioavailable copper that was in the soils used for construction. Concentrations of copper in the *C. aquatilis* at the end of 2017 were similar or lower than 2016 concentrations (Figure 22). The moss had higher copper concentrations in the A cells than in 2016, but lower concentrations in B cells, indicating more treatment earlier in the CWTS as it matures (Figure 22). Over time, constituents sorbed to mosses will form reduced minerals, rendering them less bioavailable.

Cadmium concentrations in *C. aquatilis* and moss were similar in 2017 and 2016, except for higher concentrations in cell 1B where concentrations were higher for *C. aquatilis*, (Figure 23). Similar selenium concentrations were observed in both plant types in 2017 as in 2016, (Figure 24; Contango, 2016 and 2017). Concentrations of molybdenum in moss generally decreased from 2016 to 2017, while concentrations of molybdenum in *C. aquatilis* decreased significantly and now remain around the detection limit (Figure 25). In 2017, overall concentrations of zinc decreased in *C. aquatilis* and moss from 2016, except for cell 1B in *C. aquatilis* and cell 2A in moss where concentrations of zinc remained the same (Figure 26).

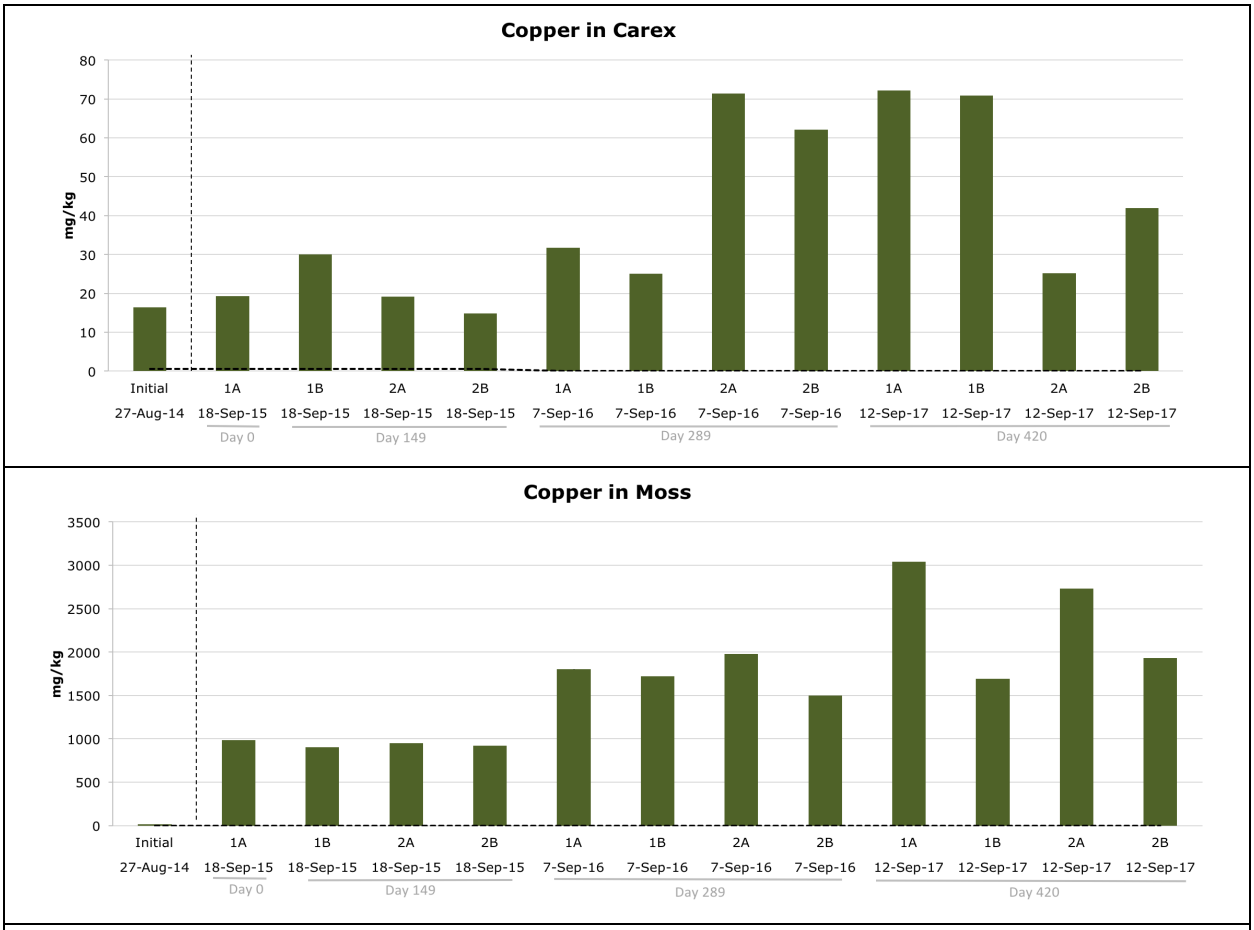


Figure 22 – Copper concentrations in plants.
The 2014 detection limit (DL; horizontal dotted line) for copper is 0.5 mg/kg, the 2015 DL is 0.1 mg/kg, and the 2016 and 2017 DL is 0.05 mg/kg. The initial data set is the average of three *Carex* replicates at construction.

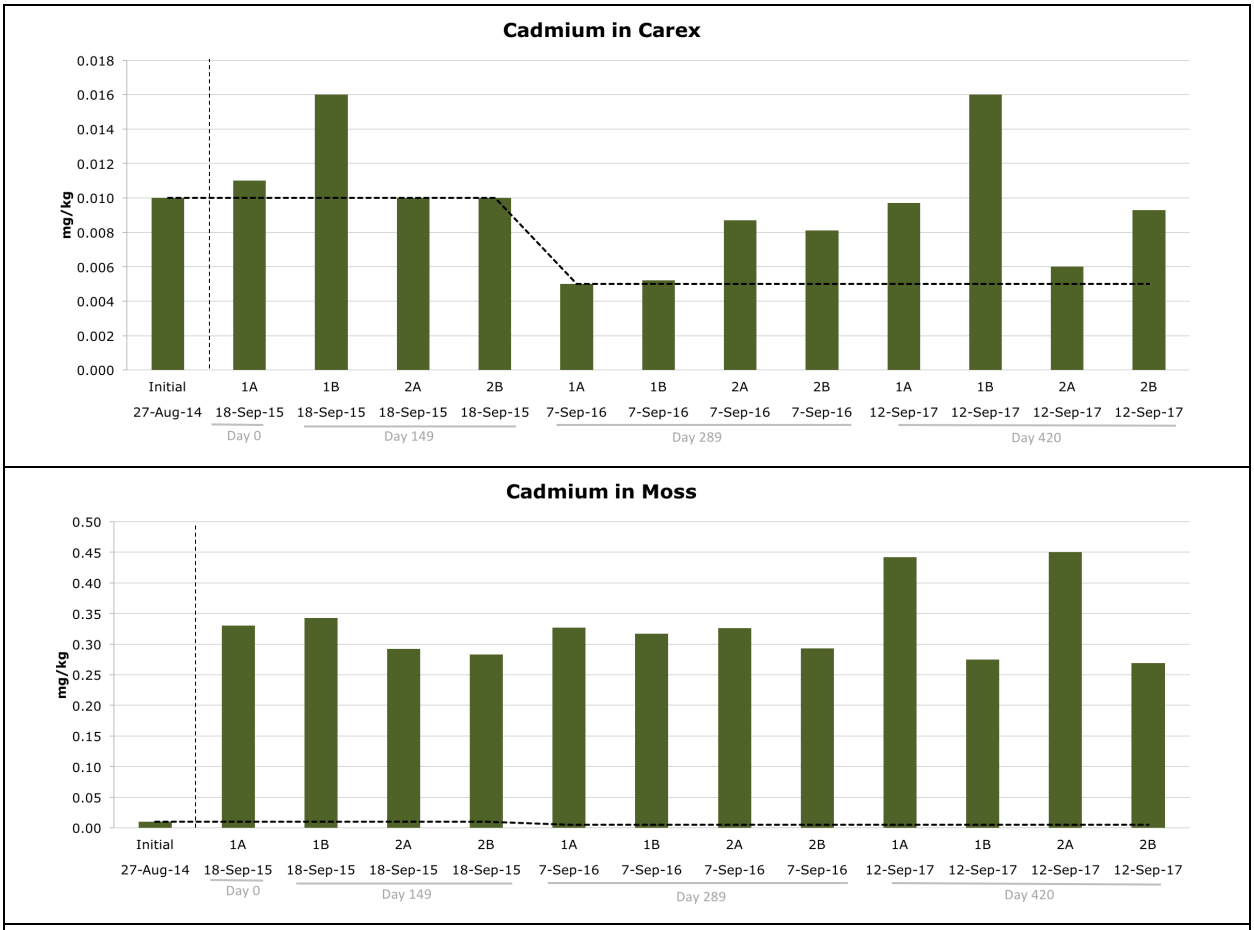


Figure 23 – Cadmium concentrations in plants.
The 2014 and 2015 detection limit (DL; horizontal dotted line) for cadmium is 0.10 mg/kg. The initial data set is the average of three *Carex* replicates at construction.

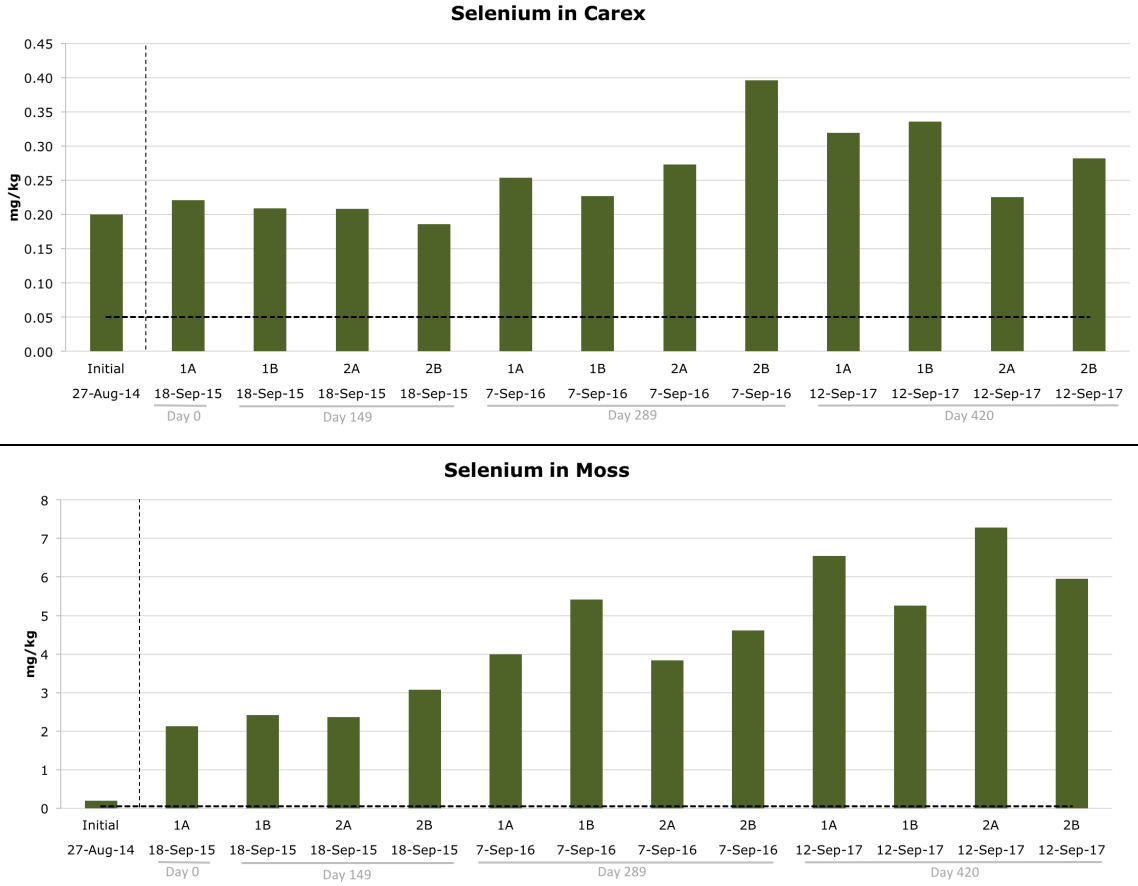
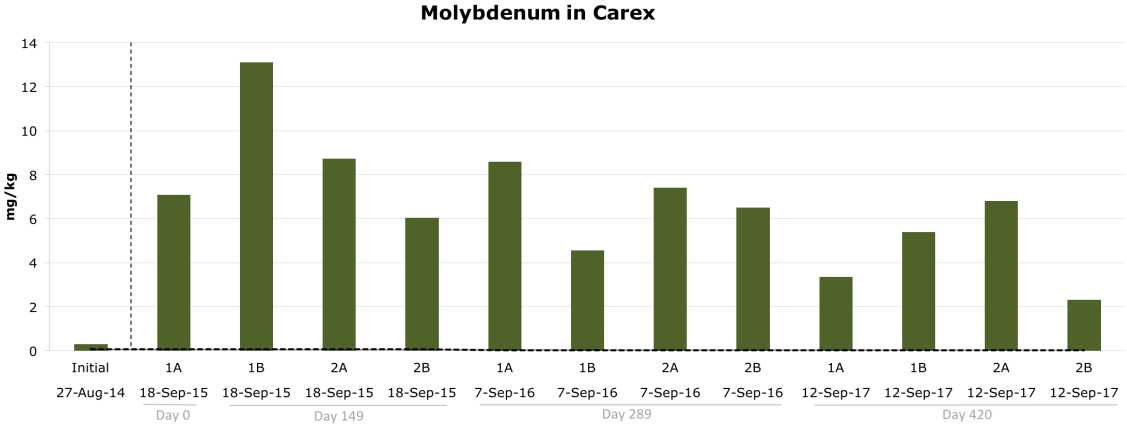


Figure 24 – Selenium concentrations in plants.
The method detection limit (DL; horizontal dotted line) for selenium is 0.2 mg/kg. The initial data set is the average of three *Carex* replicates at construction.



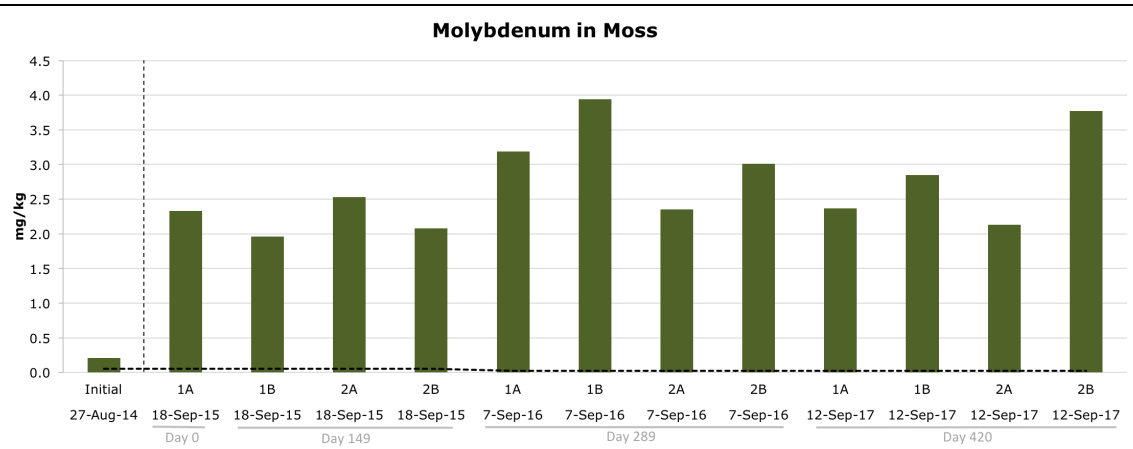


Figure 25 – Molybdenum concentrations in plants.
The 2014 and 2015 detection limit (DL; horizontal dotted line) for molybdenum is 0.05 mg/kg, the 2016 and 2017 DL is 0.020 mg/kg. The initial data set is the average of three *Carex* replicates at construction.

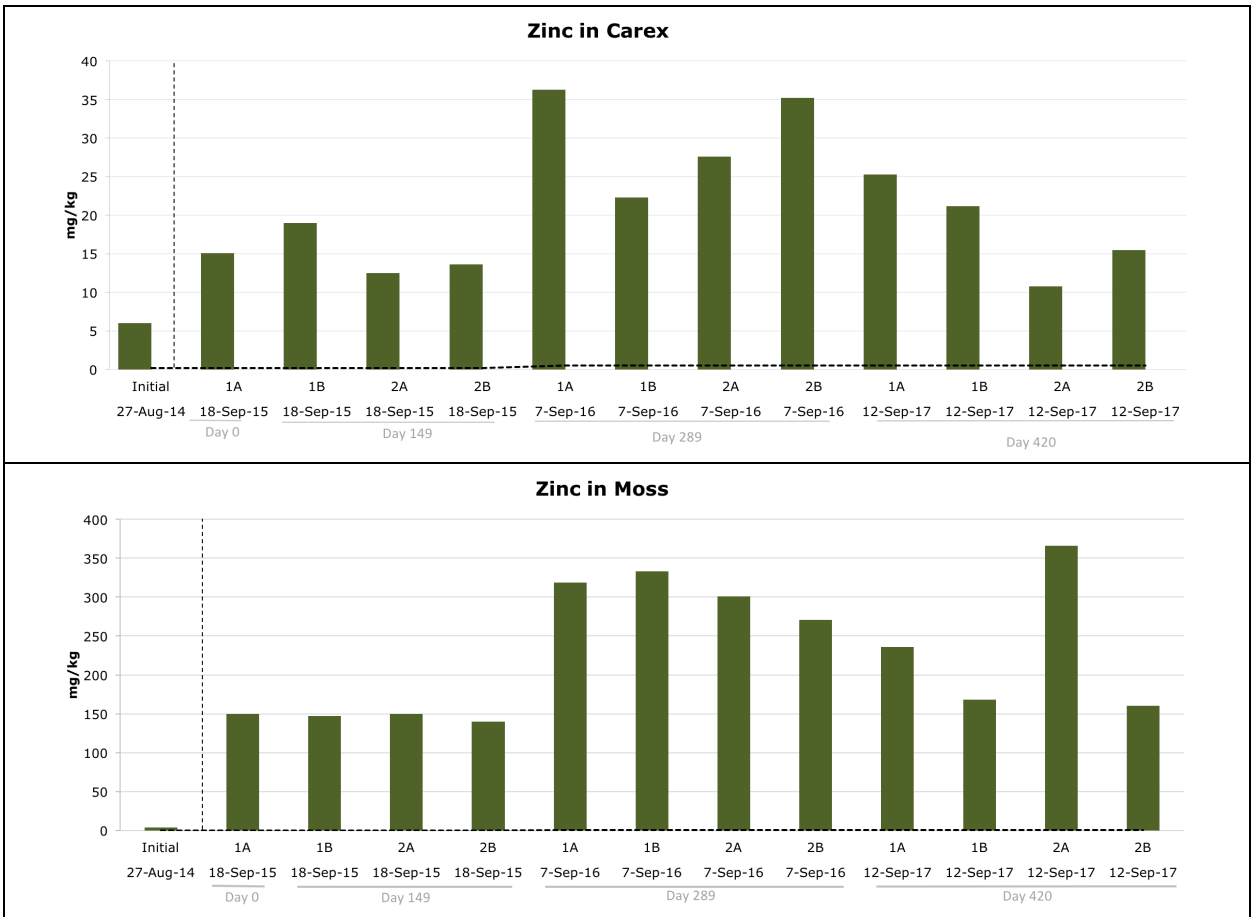


Figure 26 – Zinc concentrations in plants.
The 2014 detection limit (DL; horizontal dotted line) for zinc is 2.0 mg/kg, the 2015 DL is 0.002 mg/kg, and the 2016 and 2017 DL is 0.5 mg/kg. The initial data set is the average of three *Carex* replicates at construction.

5.8. Detritus Study

The following are key findings regarding the detritus study, which are detailed further below:

- The detritus study suggested that algae growth on the assay devices had reached a steady state (growth vs decomposition) by ~23 days of the study.
- After 83 days submerged in the CWTS, *C. aquatilis* decomposed on average 64%.

A detritus study was conducted in 2017 to assess decomposition rates of *C. aquatilis* in the CWTS over time as well as to determine the steady state of carbon contribution from algae growth on CWTS materials (Chimney and Pietro, 2006; Hammerly et al., 1989). Additional details describing the methods can be found in Appendix A. The study began on June 21, 2017 when six bags filled with 5g of oven dried *C. aquatilis* were submerged into each CWTS cell to determine the decomposition rate of *C. aquatilis* (Figure 27). Variability between samples are indicated by error bars (minimum and maximum values) in Figure 28. Six additional bags filled with 3.8 g of polyester filter fiber material were submerged into each cell of the CWTS to determine the algae growth rate. On July 25, 2017 and September 11, 2017 (after 34 and 83 days of submersion in the CWTS, respectively) one bag filled with *C. aquatilis* and one bag with the polyester fiber were sacrificed from each CWTS cell and the dried weights compared to initial sample weights (Table 9). This resulted in four replicates for each treatment and sampling date.

The weight of the polyester fiber bags increased by an average 23% after 34 days, and by 21% after 83 days, suggesting that the algae growth had reached a steady state prior to 34 days. After being submerged for 34 days in the CWTS, *C. aquatilis* decomposed on average 56% from the initial weight (decreasing from 5 g to 2.2 g), and continued to decompose to an average of 64% (decreasing from 5 g to 1.8 g) after submersion for 83 days in the CWTS (Table 9). There was little variability between cells. These detritus decomposition rates have been adjusted for by subtracting the weights of algal growth of the corresponding polyester fiber filled bag (Appendix A).

Four bags of each material remain in each cell for future sampling. The detritus study will continue in 2018 to enable monitoring and calculation of decomposition and accretion rates.



Figure 27 – Detritus study bags.
Left are mesh bags filled with sedges and polyester fiber fill stuffing. Left are mesh bags with *C. aquatilis* and control polyester prior to addition to the CWTS, middle are bags submerged in a CWTS cell and attached to depth stick, and right is a mesh bag with decomposed sedges and polyester fiber fill stuffing.

Table 9 – Results of the *C. aquatilis* detritus study.

	Initial weight of *C. aquatilis* (g)	Dry weight after 34 days (g) [2]	Decrease in weight after 34 days (%)	Dry weight after 82 days (g) [2]	Decrease in weight after 83 days (%)
1A	5.0[1]	2.1	57%	1.7	65%
1B		2.1	59%	1.7	67%
2A		2.3	54%	2.0	61%
2B		2.4	52%	1.9	63%
Average		**2.2**	**56%**	**1.8**	**64%**

[1] Initial weight prior to bags being placed in CWTS.
[2] Weights of *C. aquatilis* corrected by subtracting the increase in weight from the polyester fiber fil to account for algae growth.

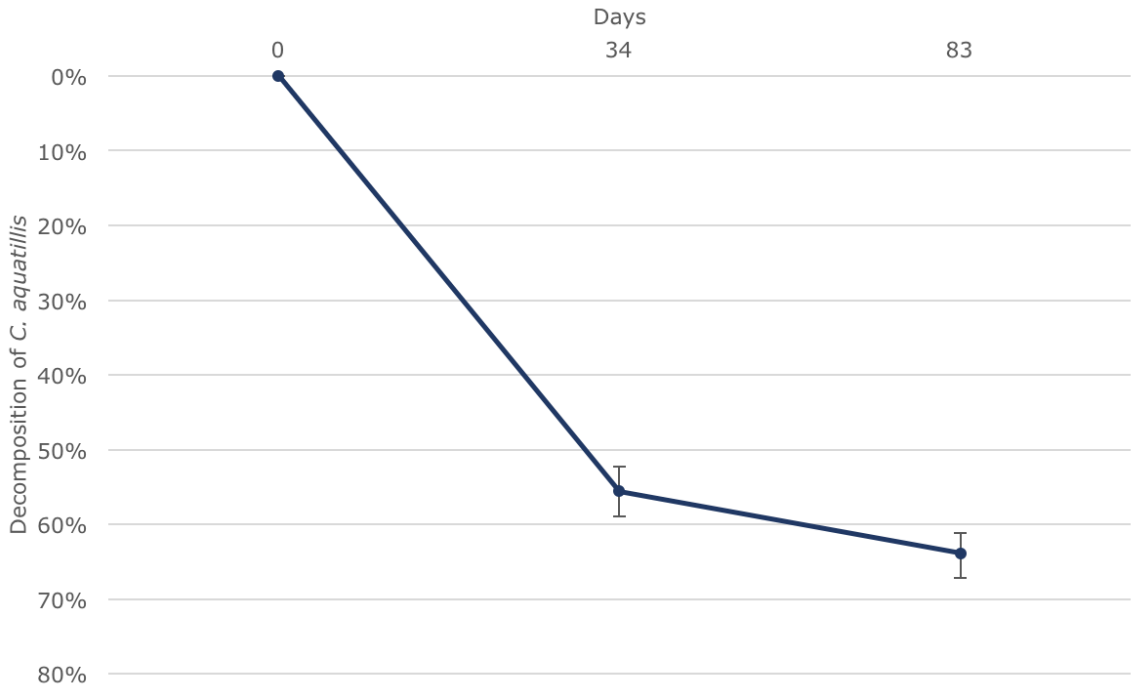


Figure 28 – *C. aquatilis* decomposition over time.
Decomposition was measured by the decrease in weight over time of the *C. aquatilis* samples compared to the initial weight. The decomposition rate was corrected for algae growth by subtracting the increase in weight of the polyester filter foam from the dry weight of the *C. aquatilis* samples over time. Error bars indicate the variation in results for each time point.

5.9. Beneficial Microbes

The following are key findings regarding beneficial microbe establishment in the CWTS, which are detailed further below:

- Establishment of sulphide-producing bacteria (SPB) increased throughout 2017 and was highest after commissioning was completed.
- SPB were found in highest abundance in root and soil samples.
- The average number of different types of sulphide-producing bacteria increased over time in all sample types tested (root, soil, detritus, and moss).
- Selenium-reducing and nitrate-reducing bacteria increased over time with the highest abundances found in *C. aquatilis* roots.

Microbes are the driving force of many treatment pathways that are targeted in CWTSs. The beneficial microbes in these CWTSs catalyze biogeochemical processes that remove specific constituents of concern from the water column. Careful design of CWTSs can create the environmental conditions needed to enhance the abundance and metabolic activity of these beneficial microbes. Accordingly, complimentary methods of genetic and growth-based testing were used to characterize the microbial populations associated with a range of microbial habitats in the demonstration-scale CWTS (e.g., soils, sediment, biofilms, aquatic mosses, and plant roots).

In the context of the Minto Mine CWTS, beneficial microbes include those that are involved in the reduction of selenium (i.e., selenate and selenite), nitrate, and sulphur compounds. Reduced sulphur can in turn treat copper, cadmium, molybdenum, and zinc through geochemical interactions. Information on each of these mechanisms and the associated microbial populations in the demonstration-scale CWTS is outlined in the following sections.

5.9.1. Sulphide-producing Bacteria

Treatment of metals and metalloids can be achieved by targeting the lithic biogeochemical sequestration of divalent metals through sulfide (i.e., S^{2-}, HS^{-}) precipitation as mineralized species (e.g., chalcocite [CuS], covellite [Cu_2S]). These sulfide-bound species are relatively insoluble (CuS; $K_{sp}=10^{-16}$; Stumm and Morgan 1996), and are transferred from the water column into the CWTS soil as non-bioavailable fractions (Murray-Gulde et al., 2003; Huddleston et al., 2008). Moreover, similar reactions occur with cadmium and zinc, rendering them non-bioavailable. As such, sulphide production is a key biogeochemical mechanism for water treatment at Minto Mine. Sulphides can be created by beneficial microorganisms through the reduction of sulphur-containing compounds, such as sulphate, sulphite, thiosulphate, and elemental sulphur.

Based on the information gathered in pilot-scale testing, the targeted soil redox for the demonstration-scale CWTS is between -100 and -250 mV to facilitate these reactions. This is in agreement with literature that indicates anaerobic conditions with relatively low ORP (-250 to -100 mV) are necessary to promote anaerobic metabolism in bacteria which oxidizes

organic matter and produces electrons, which in turn reduces sulfate to hydrogen sulfide (H_2S) and other sulfide species (i.e., bisulfide ion (HS^-), sulfide ion [S^{2-}]; Mitsch and Gosselink 2007). In these redox ranges, bacterial sulphide-production through reduction of sulphur compounds is expected, alongside increases in the proportion (percentage) and abundance of these microbes.

As expected, diversity (Table 10), proportion, and inferred abundance of sulphide-producing bacteria (SPB) increased as the soil redox decreased in the demonstration-scale CWTS with the highest proportions observed in 2017 compared to 2016, as the CWTS has matured (Figure 29, Figure 30, Figures B28 and B29 in Appendix B; Contango, 2017). Proportions of SPB increased in all sample types (soil, roots, moss, and detritus) over time through 2017 (Table 10). As observed in 2016, the *C. aquatilis* roots were again found to harbour the highest proportions and abundances of beneficial SPB, followed by the soil, detritus, and moss. Furthermore, the highest proportions were observed in the operational period for both soil and root samples. Root, detritus, and soil samples harboured the highest average number of different SPB types, followed by moss (Table 10). However, the average number of different types of SPB increased over time in every sample type tested.

The microbial analysis of various sample types in the CWTS has therefore confirmed that the commissioning period proceeded as expected and was successful in establishing beneficial SPB populations alongside reducing conditions. Monitoring of soil redox as well as microbial populations will continue to be monitored for stability or further improvement in 2018.

Table 10 – Average number of different types of sulphide-producing bacteria in 2014/2015, 2016, and 2017.

Sample Type	2014 & 2015	2016	2017
detritus	NT	6	9
moss	3	4	8
root	6	8	9
soil	4	8	9

NT – not tested. The number of different types is based on counting the number of operational taxonomic units (clustered at 97% identity) that are classified as known sulphide-producing bacteria.

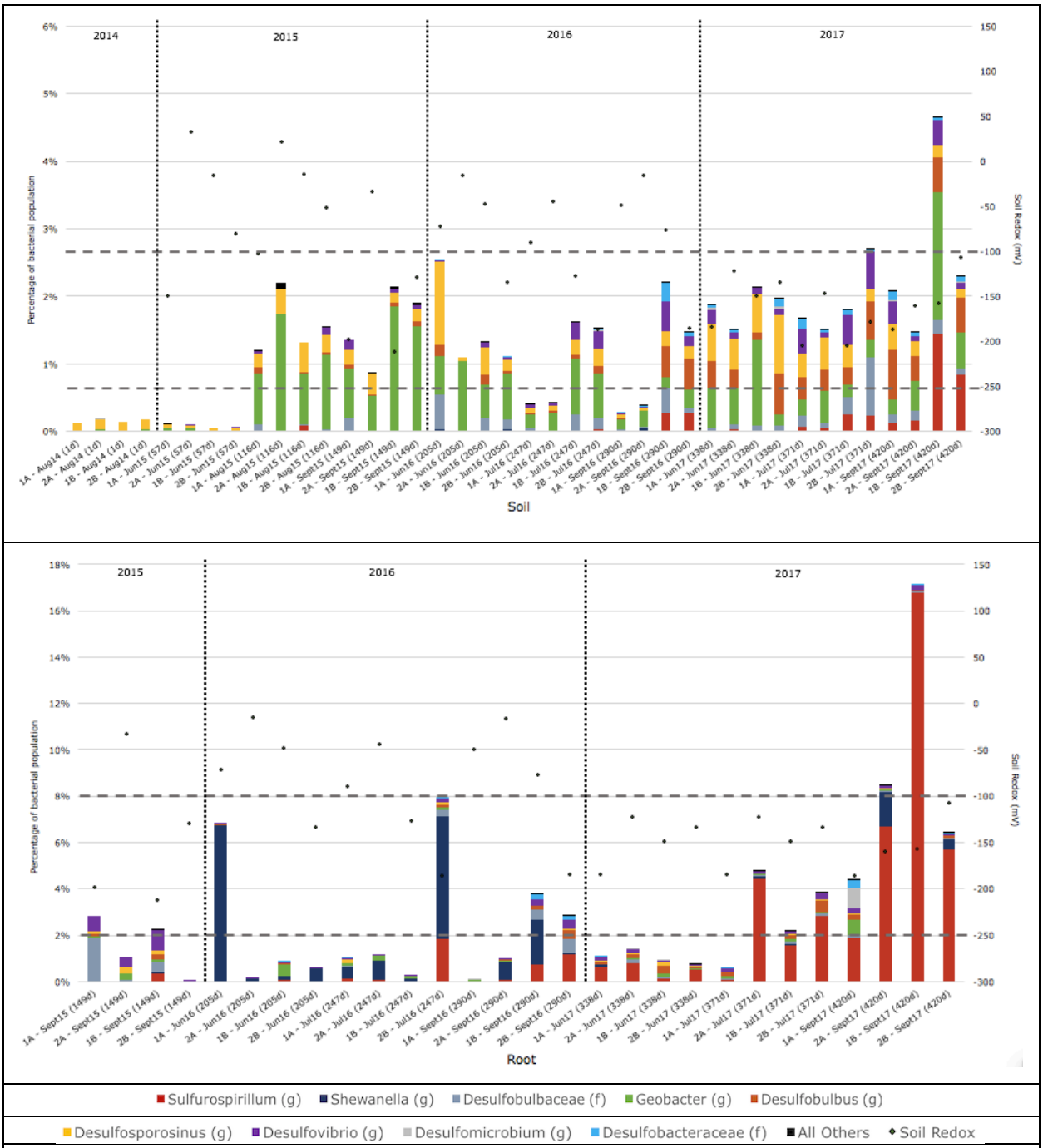


Figure 29 – Percentage and identity of sulphide-producing bacteria in soil (top) and roots (bottom) demonstration-scale CWTS.

The horizontal dashed lines indicate the targeted soil redox range. Organism classifications and data analysis methods have been updated from Contango (March 2016). The y-axis provides the percentage of the bacterial community (relative abundance) that corresponds to sulphide-producing bacteria through the reduction of sulphate, sulphite, thiosulphate, and sulphur. Percentage is based on identification via genetic sequencing. Organisms are either classified to the genus (g) or family level (f). The root y-axis is different from the soil y-axis as the % SPBs is much higher in one sample of root (1B – Sept17 (420d)).

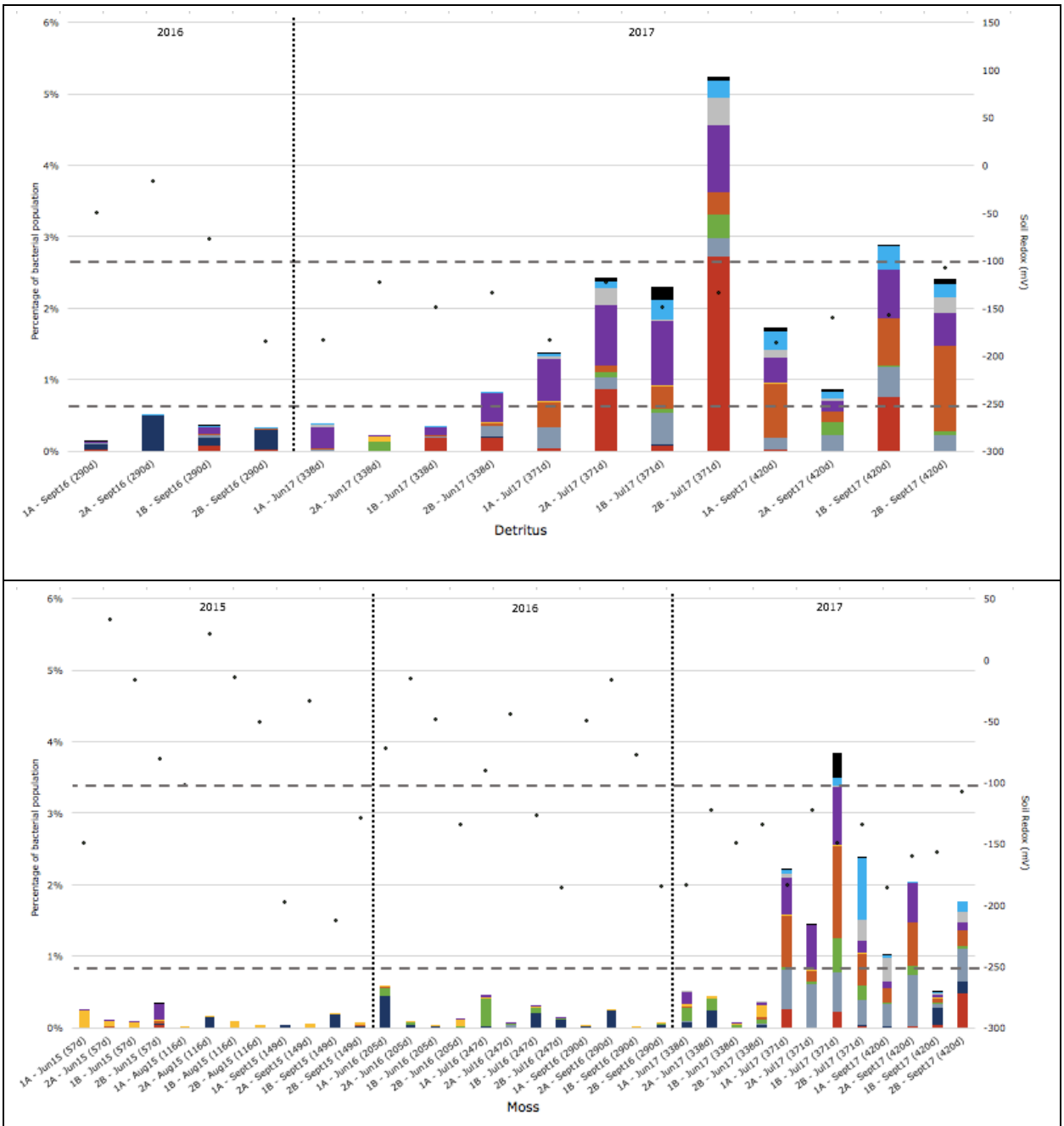


Figure 30 – Percentage and identity of sulphide-producing bacteria in detritus (top) and moss (bottom) demonstration-scale CWTS.

The horizontal dashed lines indicate the targeted soil redox range. Organism classifications and data analysis methods have been updated from Contango (March 2016). The y-axis provides the percentage of the bacterial community (relative abundance) that corresponds to sulphide-producing bacteria through the reduction of sulphate, sulphite, thiosulphate, and sulphur. Percentage is based on identification via genetic sequencing. Organisms are either classified to the genus (g) or family level (f).

5.9.2. Selenium-reducing Bacteria

The targeted selenium treatment pathways in the Minto CWTS include sorption to moss and soils, and subsequent microbial reduction of soluble (sorbed) selenate (Se(VI)) and selenite (Se(IV)) to insoluble elemental selenium (Se(0)). This reductive process can also be achieved directly in the water column, but is more effective when associated with mosses and biofilms due to their sorptive properties that bring the selenium in contact with beneficial selenium-reducing bacteria. Selenium reduction biogeochemical processes are achieved within the range of soil redox conditions targeted for sulphate-reduction as suggested by pilot-scale testing and literature (Contango, 2014).

Selenite-reducing organisms are ubiquitous in nature and as expected, were detected in all sample types, including algae, biofilm, moss, soil, sediment, roots, and detritus. Although organisms that reduce selenate to elemental selenium (rather than intermediary selenium compounds) are generally less abundant in the environment, they were found associated with all sample types, indicating that the conditions conducive to their proliferation have been created within the CWTS. Moreover, the abundance of selenite- and selenate-reducing organisms generally increased or remained stable over time in the demonstration-scale CWTS through the commissioning and operational periods in (Figure B30 in Appendix B). Aquatic mosses were found to initially host the highest abundance of both selenate- and selenite-reducing organisms, affirming the importance of the inclusion of moss in the CWTS. Over time, as the vegetation has established, selenium-reducing bacteria have increased in abundance on the roots of *C. aquatilis* (Figure B30 in Appendix B).

These findings indicate the commissioning period was successful and the demonstration-scale CWTS has established and maintained beneficial selenium-reducing microbes. Selenium is removed as it is sorbed to moss or detritus or by interacting directly with selenium in water that has been drawn into the root zone by plants. Abundance of selenium-reducing bacteria is similar to that found through 2016 and during pilot-scale soil testing, suggesting they have established as expected. Selenium-reducing microorganisms will continue to be monitored in 2018, alongside performance testing.

5.9.3. Nitrate-reducing Bacteria

Nitrate is sometimes a constituent of concern during operations and early closure owing to residuals from blasting activities. Even if not in exceedance of water quality guidelines in terms of receiving environment objectives, nitrate often requires treatment in order to subsequently achieve treatment of other constituents such as selenium and metals through sulphide production. Nitrate can be removed from water through denitrification by different types of microbes, including nitrate reducing bacteria which can reduce nitrate (NO_3) to nitrite (NO_2), and also denitrifying organisms that are capable of fully reducing nitrate to nitric oxide (NO), nitrous oxide (N_2O), and dinitrogen gas (N_2, which is the most abundant gas in air). Most-probable number (MPN) analysis was therefore used to quantify these organisms.

As observed in 2016, nitrate-reducing and denitrifying organisms were found associated with all sample types in the demonstration-scale CWTS (Figure 31; Contango, 2017). Roots and detritus had a high abundance of both nitrate and denitrifying organisms, with soil being similar to or slightly less than what was found during pilot-scale testing. These results indicate nitrate reducers have established in the CWTS during the commissioning period as expected. Nitrate-reducing microorganisms will continue to be monitored in 2018, alongside performance testing.

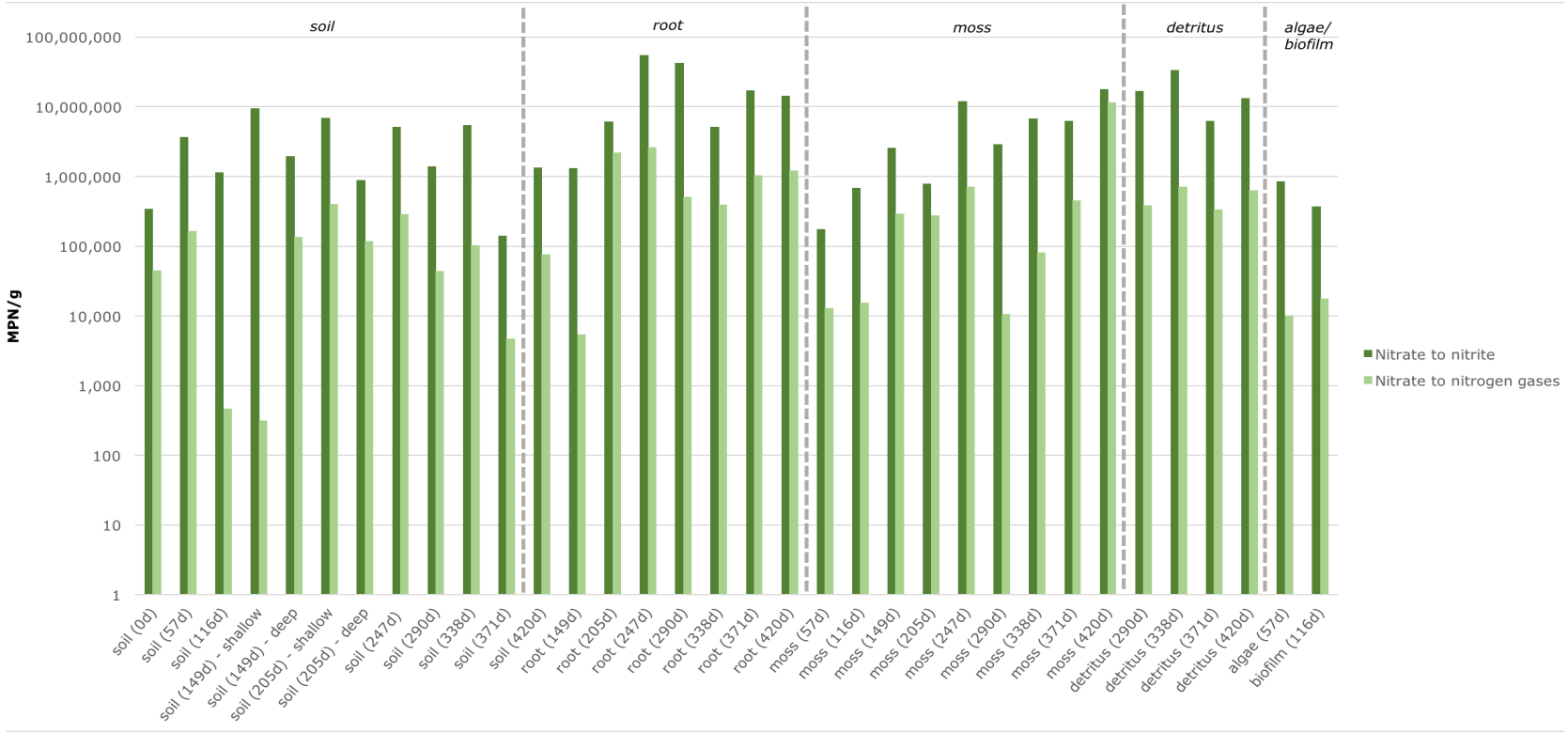


Figure 31 – Abundance of nitrate-reducing and denitrifying organisms in various CWTS sample types over time.
Nitrate-reducing = reduction of nitrate to nitrite; denitrifying = reduction of nitrate to nitrogen gases.
Shallow soil (0-5 cm); deep soil (10-20 cm); soil at 0d was that used for construction of CWTS.

6. Operational Challenges

The following are key findings regarding operational challenges, which are detailed further below:

- Aphids were again found in abundance on *C. aquatilis* in 2017.
- Efforts to control the aphid population on-site were made but an ongoing spraying regiment was not maintained and, therefore, aphids remained. However, no short-term detrimental effect on *C. aquatilis* viability was observed, despite the persistence of aphids.
- Damage to the above water vegetation was observed however new shoots were emerging in the CWTS and treatment COCs was not impacted by damage to the above water vegetation.
- Effects of aphids on *C. aquatilis* did not affect treatment.
- These issues with aphids are likely owing to the isolated nature of the demonstration-scale CWTS and not expected to affect the full-scale CWTS.

The Minto Mine's demonstration-scale CWTS, which is constructed of an emergent macrophyte monoculture of *Carex aquatilis*, experienced an aphid infestation in 2016 and 2017. The demonstration-scale CWTS is located in a relatively isolated area, away from other vegetation and potential sources of insects. It is known that aphids can migrate on air currents (Dixon, 1971), possibly over great distances (Riley et. al., 1995), which may be how they reached the demonstration-scale CWTS. Aphids are often monophagous, meaning they feed on one species of plant (Foottit and Maw, 1997) and it is likely that having additional vegetation available will not reduce the level of infestation on the *Carex* species planted in the CWTS. However, if additional surrounding vegetation is available, it could provide habitat for natural predators of aphids. With the demonstration CWTS currently located 50 m away from trees or other vegetation, there is little to no opportunity for natural predators to inhabit the area or incentive to travel the distance to the CWTS. The large buffer area (MVFE) around the demonstration-scale CWTS allows the aphids to colonize more robustly than anticipated in the full-scale CWTS onsite, which would not have the same buffer zone. This would not be an issue in the full-scale CWTS.

Aphids can be controlled, naturally, by predatory insects such as ladybugs, lacewings, and parasitic wasps (Flint, 2001). Ladybugs were observed in very small numbers on the *Carex* plants but in insufficient numbers to control the aphid population. Lacewings are found throughout North America and would be expected to be present in the Minto Mine region (Russel and Diaz, 2015). Several types of parasitic wasps that prey on aphids are known to be present in the Yukon (Finnamore, 1997). Other control methods involve the use of insecticidal soaps, which are recommended as they are non-toxic to most other insects and kill aphids through contact (the soft bodies are suffocated by the fatty acids in the soap; Ubl and Munnerlyn, 2009).

To quantify the aphid population on-site, sticky, yellow insect traps were installed at W10 and W15 as well as eight (two per cell) in the CWTS in 2017 (Figure 32). However, none of the traps at any of the locations were successful in trapping aphids and instead trapped numerous

other insects. It was observed that the sticky traps at W10 and W15 trapped a larger quantity and diversity of insects compared to the traps installed in the CWTS. This suggests that the larger population and diversity at the W10 and W15 areas assisted with keeping aphid populations under control in these areas. Due to the isolated area of the CWTS, away from other vegetated areas, the CWTS hosted a smaller quantity and diversity of insects which were not able to keep the aphid population under control.

Efforts were made in 2017 to control the aphid population by applying a mild insecticidal soap (Scott's EcoSense Bug B Gone Insecticidal Soap concentrate mixed with water) or a soap solution using a hand-held sprayer. Although the insecticidal soap was more effective than the soap solution, an ongoing spraying regiment was not maintained long enough to decimate the population. Care must be taken when using insecticidal soap to spray all surfaces of plant leaves, as aphids frequently inhabit the underside of leaves. Applications should be repeated every 4-7 days (follow label instructions) until pests are eliminated (Ubl and Munnerlyn, 2009). A stronger insecticidal soap (Trounce's Yard and Garden Insecticide) was applied on September 13, 2017 in another effort to reduce aphid populations; however, no discernable change was noticed, perhaps due to the lateness in the season. Insecticidal soaps with natural plant-based pyrethrins may provide control with fewer negative impacts than insecticides like malathion, permethrin, and acephate because pyrethrins break down quickly (Flint, 2001).

Series 1 of the CWTS appeared to have a larger aphid infestation and resulted in more damage to the above water vegetation. Regrowth will be monitored in 2018 and replanting of Series 1 may be required. The aphids will continue to be monitored in 2018 and compared with other background areas (e.g., W10 and W15) to see if increases in aphid populations are specific to the CWTS, or general to the area. In 2018, it is recommended that a stronger insecticide be applied early in the season and on a frequent, predetermined schedule.



Figure 32 - Insect traps.
Insect traps installed in CWTS cells (left and right), insect trap installed at W15 (center) in 2017.



Figure 33 – Aphids observed in cell 1A and 2A in 2016 (top) and 2017 (bottom).
Left pictures are of cell 1A. Right pictures are of cell 2A. Note the increased yellowing in cell 1A (2017) and cell 2A (2016, 2017) due to a denser aphid population.



Figure 34 – Aphids observed on *C. aquatilis.*
The left picture was taken in 2016 and the right picture was taken in 2017. Note the small green and black aphids on *C. aquatilis* leaves.

7. Summary of Results

When designed and implemented in a strategic and scientifically guided manner, CWTS can mitigate risks posed by many constituents. A treatment plan including processes to precipitate insoluble species of these constituents for sequestration into the soils of the wetland are very desirable as this mechanism captures the constituents and stores them in stable form in the soil, rather than transferring the constituents to an indeterminate fate (e.g., through plant uptake that can potentially bio accumulate in wildlife or be re-released in plant decomposition). This study addressed several important design considerations regarding implementation of a CWTS at full-scale for the treatment of metals and metalloids to meet the overriding objective of completing commissioning of the CWTS and progress through operational performance. A summary of key findings from the 2017 studies and recommendations to meet the objectives are listed below.

Monitor explanatory parameters and performance to determine when commissioning is complete and the operational period has begun:

- Dissolved oxygen (DO) decreased from an average of 8.4 mg/L during commissioning-B in 2016, to an average of 5.3 mg/L during operations in 2017. The DO in the water column is likely the result of photosynthesis of algae and mosses.
- Despite this DO level in the water column being in oxidizing ranges, stable reducing conditions were achieved in the CWTS soils within the targeted soil redox range (-100 to -250 mV).

Assess removal of constituents from the water:

- Copper treatment in the CWTS was masked by leaching from the soils used in construction of the CWTS into the water, but this has mostly been remedied now by the wetland treating this copper and turning it into more stable sulphide forms in the soil.
- During the operational period the demonstration-scale CWTS successfully achieved an average decrease in concentrations of 0.0169 µg/L for cadmium (from 0.0261 µg/L to 0.0092 µg/L), 31.8 µg/L for copper (from 49.1 µg/L to 17.3 µg/L), 3.6 µg/L for molybdenum (from 6.3 µg/L to 2.7 µg/L), 3.5 µg/L for selenium (from 4.0 µg/L to 0.5 µg/L), and 47.3 µg/L for zinc (from 49.2 µg/L to 1.9 µg/L).
- Molybdenum and selenium treatment in the operational period is notable as the removal rates were negligible within the margins of error of the testing method in the commissioning-A period.

Determine the hydraulic residence time (HRT) by tracer study and associated correction factor to apply to the nominal (calculated) HRT:

- The tracer study effectively demonstrated the HRT (2.25 days) and flow symmetry through the CWTS.
- There was a single flow path in the CWTS (shown by a single peak in the tracer study)
- Water is incorporating into the CWTS soils (shown by the long tail for depletion of the tracer).

- The nominal HRT is calculated from the area of the CWTS and the depth at the in-situ measuring points. This nominal HRT does not account for depth variations, embankment slopes, vegetation (using space in the water), or substrate pore space involvement. It was found that once all of these factors are in play, the correction factor from nominal to actual is only 0.01 added to the depth of the CWTS, which is incorporated into the HRT calculation as expressed in Equation 3.

Evaluate CWTS performance, and determine achievable concentrations of contaminants of concern (thermodynamic minimums):

- All targeted constituents are being treated by mineralization and sequestered to the soils (minimal plant uptake).
- The lowest concentrations consistently achievable for the treatment design (thermodynamic minimums) were reached by the end of the A cells for cadmium and copper.
- RRCs for cadmium and zinc in the 2017 demonstration-scale CWTS were artificially low because low flow rates did not provide the resolution needed to determine a RRC.
- Removal rate coefficients (RRCs, *k*) have been developed that can be used for full-scale sizing.
- Copper leaching from soils has decreased, but is still likely making the RRC artificially low in this CWTS; however, the RRC is expected to improve once copper leaching has subsided.

Update site-specific removal rate coefficients (from commissioning period) with data from operational period:

- Removal rate coefficients (RRCs, *k*) have been developed that can be used for full-scale sizing.
- Copper leaching from soils has decreased, but is still likely making the RRC artificially low in this CWTS.

Determine amount of water loss due to evapotranspiration and effect on outflow concentrations:

- The evapotranspiration studies revealed a significant loss of water, which will impact calculations of loads to the receiving environment (making them lower than previously estimated).
- In May and June, an average water loss of 5.3 L/day/m^2 was observed, which is equivalent to 18-20% of water (~700 L/day lost in the demonstration CWTS).
- During the evapotranspiration trials, copper leached into the water as it was transformed from an oxide mineral to a sulphide mineral (because of the copper in the soils used for construction). This is not representative of what would occur during periods with no flow in a full-scale CWTS, where soils with minimal leachable copper are used and copper is deposited in sulphide form (fraction 4) by the biogeochemical activity of the CWTS (Section 5.6).

Monitor metals leaching from mineralized soils used in construction:

- In 2017, leachable copper concentrations in soils decreased in the top 0-10cm while total copper concentrations increased.

Assess stability of constituents of concern in soils:

- Most constituents, including copper, have shifted primarily into stable reduced and residual minerals fractions in the soil.
- Acid volatile sulphides (AVS) were non-detectable in the CWTS in 2016. In 2017, small amounts of AVS were detected in cells 1A and 2A which indicates that residual sulphides are starting to become available for metal treatment and that copper in the soils are becoming rendered inert through sulphide mineralization.

Determine the rate and extent of detritus decomposition (*C. aquatilis* leaves) in the CWTS over time:

- The detritus study suggested that algae growth on the assay devices had reached a steady state (growth vs decomposition) by ~23 days of the study.
- After 83 days submerged in the CWTS, *C. aquatilis* decomposed on average 64%.

Assess treatment mechanisms (including microbes):

- Establishment of sulphide-producing bacteria (SPB) increased throughout 2017 and was highest after commissioning was completed.
- SPB were found in highest abundance in root and soil samples.
- The average number of different types of sulphide-producing bacteria increased over time in all sample types tested (root, soil, detritus, and moss).
- Selenium-reducing and nitrate-reducing bacteria increased over time with the highest abundances found in *C. aquatilis* roots.

Determine an appropriate method for insect pest control (aphids) in the CWTS:

- Aphids were again found in abundance on *C. aquatilis* in 2017.
- Efforts to control the aphid population on-site were made but an ongoing spraying regiment was not maintained and, therefore, aphids remained. However, no short-term detrimental effect on *C. aquatilis* viability was observed, despite the persistence of aphids.
- Damage to the above water vegetation was observed however new shoots were emerging in the CWTS and treatment COCs was not impacted by damage to the above water vegetation.
- Effects of aphids on *C. aquatilis* did not affect treatment.
- These issues with aphids are likely owing to the isolated nature of the demonstration-scale CWTS and not expected to affect the full-scale CWTS.

8. Next Steps for 2018

A conceptual testing plan was developed at the beginning of the demonstration-scale CWTS program, and has been refined and adapted based on performance and scientific findings during the commissioning and operational periods of the demonstration-scale CWTS. The demonstration-scale CWTS is expected to run until at least the end of 2018 to assess performance under a wider range of conditions. In 2018, a flow rate schedule will be developed targeting different HRTs based on predicted HRTs for full-scale performance. The overriding objective of the demonstration-scale program in 2018 is to assess performance under conditions that would be similar to the full-scale CWTS. Table 11 outlines the action plan for CWTS optimization and testing for 2018 based on the performance results through 2017.

A multi-year plan is provided in Appendix C, and includes work performed to date as well as a schedule of activities for 2018.

Table 11 – Minto 2018 CWTS demonstration-scale action items.

Timing	Task	Purpose
April 2018	Develop a flow rate schedule for 2018.	The flow rates will fluctuate based on expected amounts of water for full-scale wetland (scaled to size) in order to evaluate the performance of the wetland at different key periods.
	Develop sampling plan for 2018	The sampling plan will include sampling types, parameters, frequency, and locations. These results will inform the performance of the CWTS. Monthly sampling of the in, mid, and out locations will be added to the 2018 sampling plan to better assess removal of constituents through the CWTS cells.
	Develop aphid control and monitoring and plan for 2018	A consistent control and monitoring plan will aid efforts to control the aphid population residing in the demonstration-scale CWTS.
April/May 2018	Start water flow to CWTS	Begin 2018 demonstration-scale CWTS program
	Add sandbags on top of existing sandbags along edges of CWTS	Adding sandbags along edges of CWTS will minimize flow short circuiting. Use the sand that was confirmed to be a good borrow source in 2016 (i.e., low copper)
	Add sandbags to the end of the CWTS cells to increase the water depth to ~20 cm	Raising the water levels will promote the desired reducing conditions in the CWTS.
	Implement aphid control using insecticide on a routine schedule	Begin applications in early May when vegetation begins to become green, in order to control the aphid population before an infestation occurs.
Ongoing through 2018 sampling season	Routine sampling program	Follow sampling plan from May to September 2018 for routine collection of water, soil, microbial, plant, and detritus samples.
	Monitor for the presence of aphids in the CWTS and surrounding areas (W10 and W15 areas).	Determine if the aphid presence is isolated to the CWTS. Determine if more aggressive control measures are needed.

July/Aug 2018	Conduct a vegetation plot harvest. Collect all above water biomass of *C. aquatilis* in a 1m x 1m plot	To determine quantity of biomass per m^2
	Conduct evapotranspiration study	Although 2017 data suggests minimal release of metals from soils, another evapotranspiration trial during the operational period needs to be conducted to confirm metals leaching has subsided. Completing the evapotranspiration study in July/Aug in 2018 will provide information on evapotranspiration rates in warmer months and the effects of seasonality.
Sept 2018 (If needed)	Replant CWTS	Replant areas of the CWTS that were impacted by aphids if needed.
December 2018	Report	2018 Update Report

9. Closure

We trust the information herein satisfies your present requirements. Should you have any questions, please contact the persons listed below. We appreciate the opportunity to provide the services detailed in this report, and look forward to discussing any comments you may have.

Respectfully submitted,

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APPENDIX A - Methods

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A1. Demonstration-Scale Constructed Wetland Treatment System Design

A1.1. System Layout and Dimensions

The demonstration-scale CWTS includes two systems in parallel with two cells in each series and a final catchment basin that both systems flow into (Figure A1). Dimensions and construction details are available in the Minto Demonstration Scale Report Document 011_0315_01A (Contango, 2015). The two parallel systems serve as a replicate for data analysis, and as testing has progressed, the two systems have also allowed for comparison of different management techniques. Dimensions of the systems are provided here in Table A1.

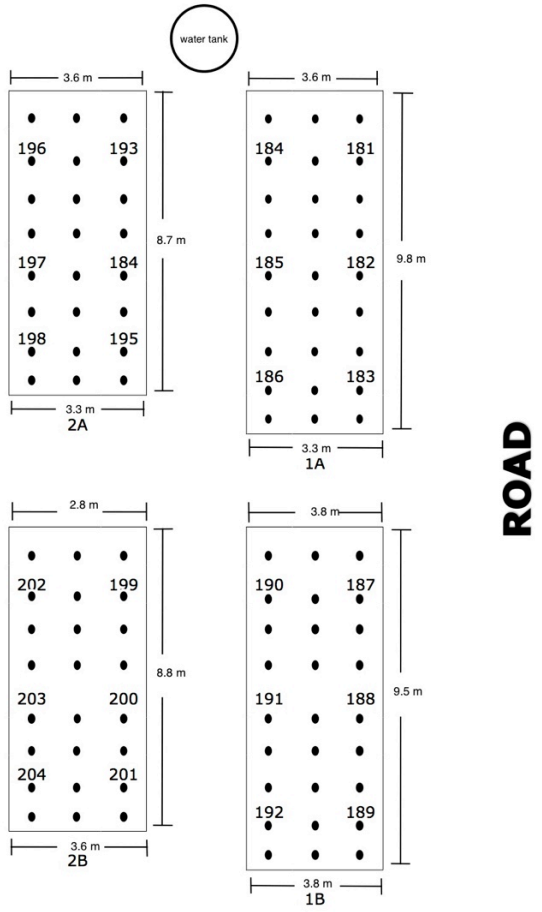


Figure A1 – Diagram of demonstration-scale CWTS.
Dimension measurements are indicated at soil surface. Black dots indicate initial construction grid marked by moss stakes, and locations of soil redox probes with identifying numbers.

Table A1 – Dimensions of demonstration-scale CWTS cells at soil surface and resultant areas of treatment systems.

Measurement		Series 1		Series 2	
		1A	1B	2A	2B
Width (m)	Inflow	3.6	3.8	3.6	2.8
	Outflow	3.3	3.8	3.3	3.6
Length (m)		9.8	9.5	8.7	8.8
Approximate surface area at soil (m^2)		33.8	36.1	30.0	28.2
Total area of system at soil (m^2)		69.9		58.2	

A1.2. Soils

Soils used to construct the CWTS are described in the initial report that outlines construction (Contango, 2015). In brief, the recommended soil for the CWTS was sand, with 2-7% by volume as organic material (e.g., woodchips, peat). In the pilot-scale systems, this resulted in a total organic carbon (TOC) content by weight of 0.2-0.6% (the sand itself was at 0.1% TOC prior to adding amendment). Ideally, this percentage could be higher, approximately 2-10% by weight to stimulate the desired reducing conditions. For the demonstration-scale system, the soil added to each of the four cells during construction was from a local borrow site. As is expected in a mining area, the soils are likely mineralized. Although a potential borrow source was tested prior to construction ("Tested Soil"; Table A2), a different borrow source was available upon construction of the demonstration-scale CWTS ("Soil used"; Table A2). The material used in the construction of the demonstration-scale CWTS was an organic peat, and analyses received after construction indicated an elevated concentration of leachable copper (Table A2). It should be noted that in a full-scale system, the variability in soil would be normalized by the larger volume of soil used. The soil in the CWTS (composed of the organic peat soil, wood chips, and straw) had a total organic carbon (TOC) content of 1.8-3.1%.

Table A2 – Comparison of copper concentrations in soils of pilot- and demonstration-scale CWTS.

Test method	Pilot-scale	Demonstration-scale	
	Initial soil	Tested soil (June, 2014)	Soil used (August, 2014)
SPLP Copper (mg/L)[1]	-	0.00546-0.0296[2]	0.148-0.608
Total Copper (mg/kg)	5.3-5.5	210-1400[2]	960-1400[3]

[1] SPLP - Synthetic Precipitation Leaching Procedure.
[2] For the June 2014 samples, the soil with the highest total copper concentration also had the lowest leachable copper concentration, and was therefore deemed acceptable for use.
[3] Total copper value for soils used was taken in June 2015 (no data for August 2014).

A1.3. Vegetation

The demonstration-scale CWTS was planted with *Carex aquatilis* (aquatic sedge) and aquatic mosses from the W10 area of the Minto Site. The plant selection and borrow source was previously determined through the site assessment (Contango, 2014a). Five *C. aquatilis* plants were planted per square meter, with moss tied to stakes that outlined the 1 m x 1 m grid for planting (details provided in Contango, 2015).

A1.4. Water Source

Water from the W36 area receiving seepage from the toe of the Mill Valley Fill Extension (MVFE) was selected for the demonstration-scale CWTS testing as the leachate is similar to that expected upon closure in the MVFE area. The chemistry of this water at the time of bringing the demonstration-scale CWTS online (September 18, 2014) is provided in Contango, 2015. In 2015 the water for the CWTS was pumped out of the sump at the toe of the MVFE. As the Mill Valley Fill Extension Stage 2 (MVFE2) progressed a new collection sump (W62 sump) was constructed as a replacement for the sump used in 2015. The W36 sump was decommissioned in February 2016 as part of the MVFE2 construction. The new sump approximately 30 m downslope was the supply water for the CWTS for the summer of 2016 and was the same source for 2017 (Figure A2).



Figure A2 – Water source for the demonstration-scale CWTS in 2017.

A2. Flow Rates

Flow rates as well as the totalizer values were recorded from the flow meter that feeds the demonstration-scale CWTS periodically throughout 2017. The flow rates (recorded by Minto) represent the flow that occurred at the time the flow rate was recorded off the flow meter and is represented by blue bars in Figure A3 below. The totalizer value was also recorded periodically and is the cumulative amount of water that passes through the flow meter between two given dates. Actual flow rates were also calculated based on the totalizer values and are represented by green dots in Figure A3 below. The flow rates calculated from the totalizer values represent the actual flow rates. The average actual flow rates calculated from the totalizer values during the operational period were 0.38 gallons per minutes (GPM) for series 1 and 0.29 GPM for series 2. Figure A3 shows the fluctuation in actual versus targeted flow rates throughout the 2017 operational period.

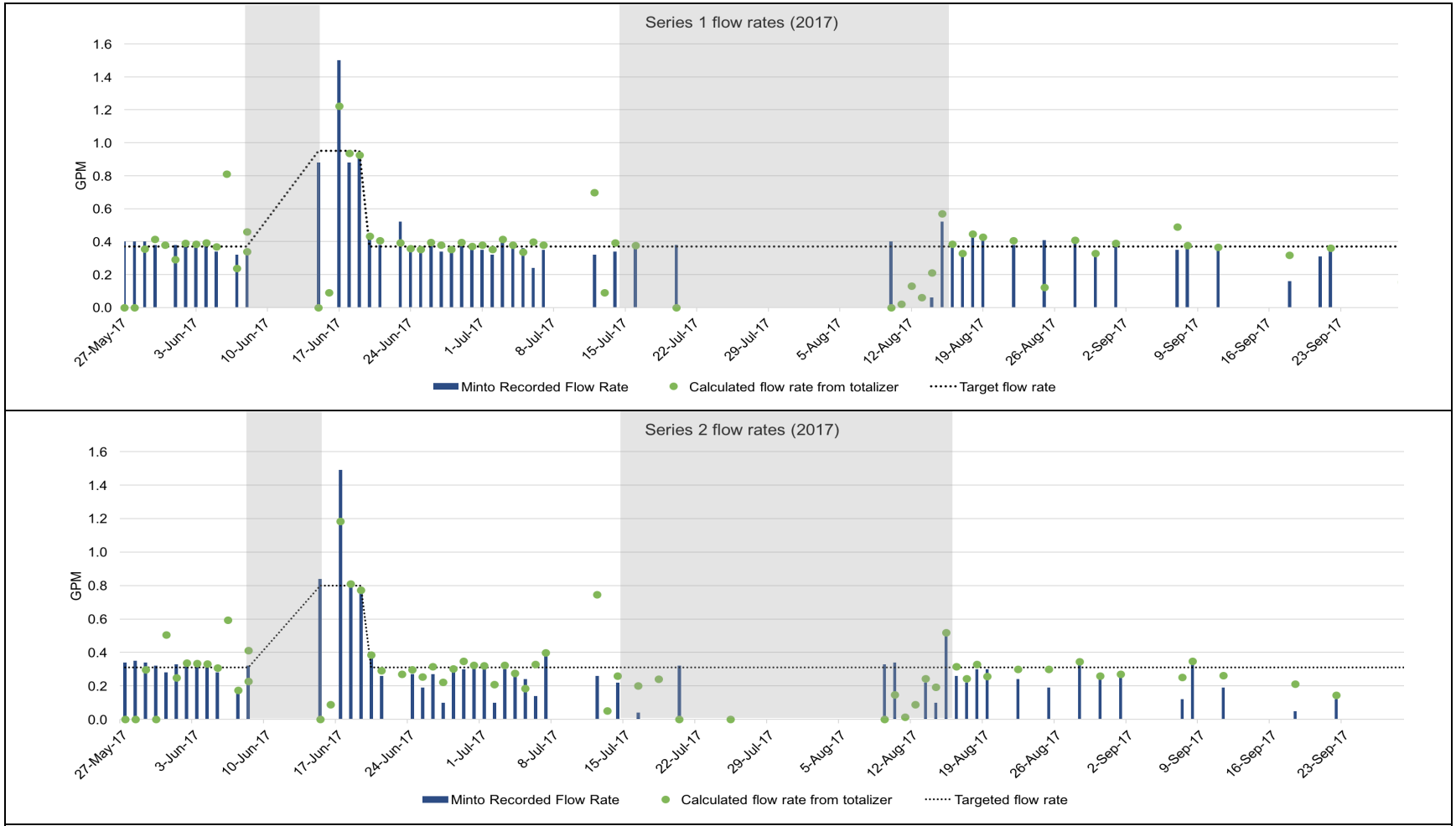


Figure A3 – Flow rates during CWTS operations in 2017.
Blue bars indicate flow rates displayed on flow meter, while green dots are flow rates calculated from the totalizer values (values from meter were recorded by Minto personnel). Targeted flow rates for series 1 and 2 were 0.37 and 0.31 gallons per minutes (GPM), respectively. Targeted flow rates are different for each series because the series differed in size. Series 2 was smaller than series 1 and therefore needed a slower flow rate to obtain the same HRT as series 1. The totalizer value is the cumulative volume of water passing through the flow meter between two measurement dates. Areas with no blue bars or green dots are time periods where flow rates and totalizer values were not recorded. Areas shaded grey indicate dates of interrupted flow. Further details can be found in Table 3 of the main report.

A2.1. Demonstration-Scale Commissioning Periods

Performance monitoring continued during commissioning although full treatment functionality of the demonstration-scale system was not expected. The information aided in guiding future development of the full-scale system in the following ways:

- Evaluated construction effectiveness and potential optimizations.
- Assessed timelines to reach targeted operational parameters to allow for effective phasing of implementation (e.g., soil redox, percentage and abundance of sulphate reducing bacteria).
- Assessed the effectiveness of *Carex aquatilis* and aquatic moss transplantation to assess planting density, time period to full density, and if plant propagation and/or a replanting schedule is necessary.

However, there were aspects of the demonstration-scale CWTS that were not directly transferrable to the planned construction of the full-scale CWTS. Notably, the demonstration-scale CWTS was built with a different soil type and chemistry than recommended for the full-scale system. Therefore; it took longer to commission the demonstration-scale system as the copper in the soils needed to be treated in addition to the copper and other constituents in the water.

Based on pilot-scale testing (Contango, 2014b), the estimated commissioning period for the Minto demonstration-scale CWTS was four months (i.e., period when water is flowing). In 2014, the demonstration-scale CWTS operated for less than one month prior to freezing, and was restarted along with spring thaw (freshet) in mid-May 2015. Based on these timelines, September 2015 was month four of commissioning, and the demonstration-scale system trended towards performance as expected during the commissioning period (Contango, 2016). However, as discussed further in sections 2.2 and 6.1 of the 2016 update report (Contango, 2017), the soil used in the CWTS had high concentrations of leachable copper, and therefore optimization opportunities were evaluated throughout 2015, and adjustments were implemented in 2015 and 2016. The last operational adjustment during the commissioning period was the addition of organic material in 2016, which marked the end of the commissioning-A period and the beginning of the commissioning-B period. Organic material will also be added during construction of the full-scale design to enhance desired reducing conditions to establish in the CWTS until vegetation is sufficient to provide for this required component. The end of the commissioning-B period was expected to be between July 16 and August 2, 2017. However, due to issues with the totalizer, the end of the commissioning-B period was August 17, 2017 and was marked by stabilization of flow rates and resolution of totalizer complications. Additionally, sandbags were added to the end of each cell on August 11, 2017 in order to raise the water level to obtain better treatment through enhancing reducing conditions. The operational period therefore began on August 18, 2017 and ran until September 22, 2017. Discussions in this report are based on the operational period.

A3. Sampling and Analyses

A3.1. Routine Monitoring of Explanatory Parameters

Explanatory parameters are quantifiable aspects of a CWTS environment that can be used to assess feasibility of treatment for a range of constituents, and therefore 'explain' the performance of a CWTS. These parameters, which often include acidity, alkalinity, conductivity, dissolved oxygen (DO), pH, oxidation-reduction potential (ORP), ion balance, available electrons donors (e.g., organic carbon, reduced elements), and temperature, can be used to predict, promote, and/or optimize the ability of the system to treat different constituents (Haakensen et al., 2015). A YSI ProPlus meter was used in the field to test for water temperature, DO, conductivity (and specific conductivity; SPC), pH, and ORP. In situ field readings were taken weekly in conjunction with water sampling using the YSI ProPlus.

A3.2. Water Sampling and Analyses

Water sampling occurred on a weekly basis from spring thaw in May 2017 to freeze up in September 2017. Frequencies and sampling parameters are outlined in Table A3. Grab samples were collected at the feed and outflow of each cell. Care was taken to avoid collecting plant matter, invertebrates, and other debris in the grab samples. Samples were collected from downstream to upstream to ensure that water sampling did not affect sample collection from the next cell (e.g., disrupting flow rates or suspending solids).

Due to copper and aluminum being found to be released from the CWTS substrates into the water in 2015, an additional sampling structure was designed. For two timepoints in 2015 (August 15, 2015 and September 17, 2015), two timepoints in 2016 (June 14, 2016 and September 7, 2016), and three timepoints in 2017 (June 20, 2017, August 18, 2017, and September 11, 2017) water was sampled not only at the feed and the outflow of each cell, but also through the cell. Beginning at the outflow of cell B and working upwards towards the feed, samples were taken at the B cell outflow, B cell mid-point, and B cell inflow, and then the same three points for the A cell (Figure A4). In each case, the sample was taken from within reaching distance of the side shore to ensure that sediments were not suspended in sampling.

Table A3 – Summary of 2017 analytical water sampling types, frequencies and locations.

Water	
Dissolved and total metals	Weekly, outflow of each cell and feed
Ammonia	
Nitrate	
Nitrite	
Flow rate	Weekly, feed
pH, DO, ORP, Conductivity (in situ)	Weekly, all cells and feed
Anion Sum	Monthly, outflow of each cell and feed
Cation Sum	
Hardness ($CaCO_3$)	
Ion Balance	
Total Dissolved Solids (TDS)	
Chloride (Cl)	
Sulphate (SO_4)	
Nitrogen (Total)	
Alkalinity	
pH	
Conductivity	
Chemical Oxygen Demand (COD)	
Total Organic Carbon (TOC)	
Total Kjeldahl Nitrogen (TKN)	
Total Suspended Solids (TSS)	
Dissolved Organic Carbon (DOC)	
Bicarbonate (HCO_3)	
Carbonate (CO_3)	
Hydroxide (OH)	

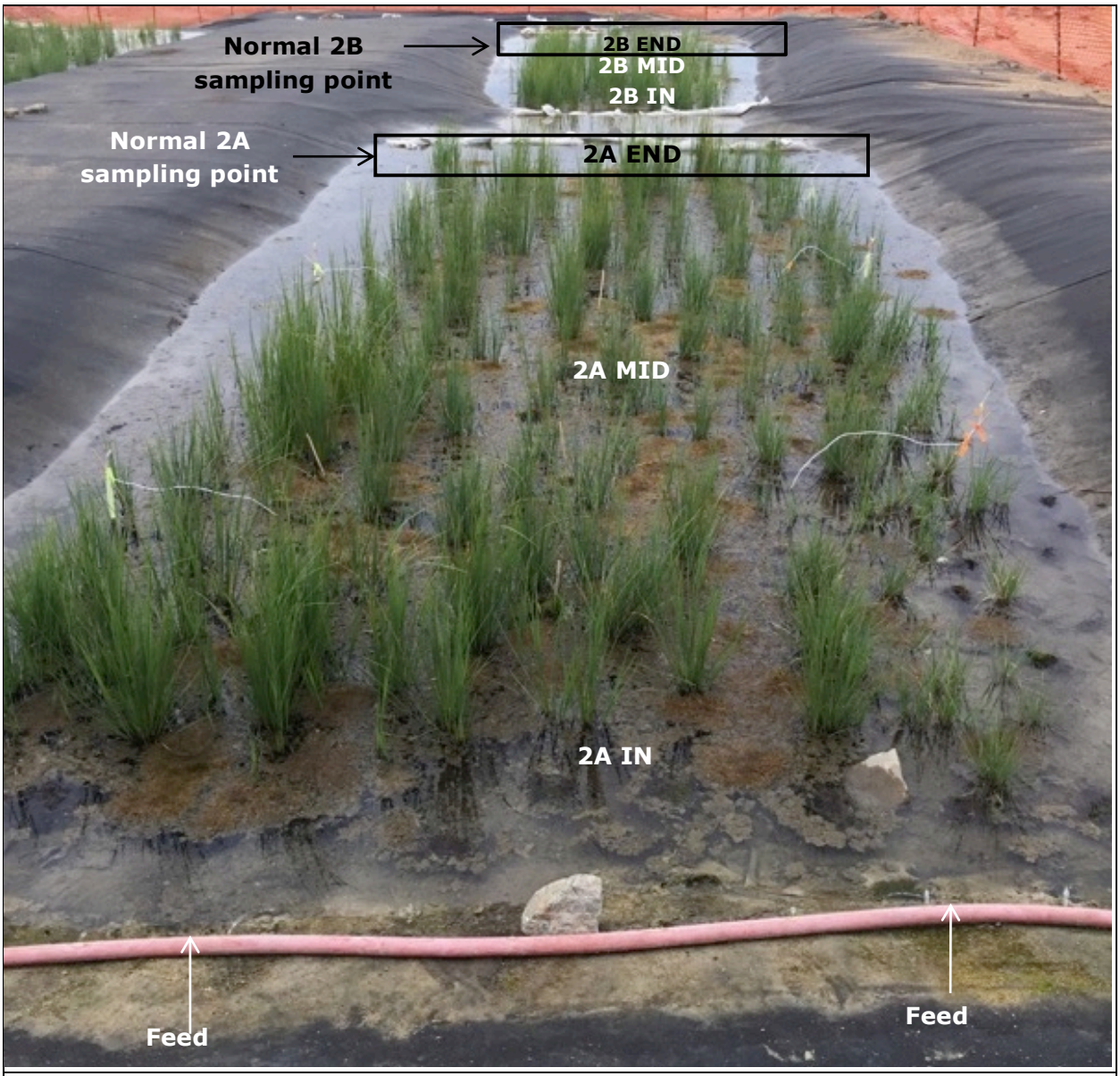


Figure A4 – Sampling points through the CWTS (Photo taken August 16, 2015).

A3.3. Soil Sampling and Analyses

Soil sampling was conducted three times in 2017 (June, July, September). A sample of the top 0-10 cm of CWTS soil was collected into a 1L glass jar for analyses. Analyses are outlined in Table A4. Each CWTS cell was sampled during each sampling event.

Table A4 – Summary of 2017 analytical soil sampling types, frequencies and locations.

Soil	
Relative soil redox (in situ)	Monthly, all probes (6 per cell)
Analyses completed on soil sample	
SAR, pH, EC, %sat, Ca, F, Mg, Na, K, Cl, SO4	Seasonally (3x per year) from a depth of 0-10 cm.
Available NPK and sulphur	
Alkalinity	
Bicarbonate (HCO_3)	
Carbonate (CO_3)	
Total Organic Carbon (TOC)	
Metals Analysis (Total)	
Sulfur by LECO	
Metals Leach	
AVS+SEM	
Sequential Leaching (5 Acid Test)	
Metals SPLP	Spring Sampling (1x per year).
Analyses completed on special leach	
Bromide (Br)	Seasonally (3x per year)
Chloride (Cl)	
Sulphate (SO_4)	
Fluoride (F)	
Alkalinity	
Bicarbonate (HCO_3)	
Carbonate (CO_3)	
Hydroxide (OH)	
Dissolved Organic Carbon (DOC)	
Ammonia	
Nitrate	
Nitrite	
Total Dissolved Solids (TDS)	
Anion Sum	
Cation Sum	
Cation/EC ratio	
Ion Balance	

Although unintentional, the use of soils in the CWTS with high initial leachable copper concentrations (Table A2) allowed for additional types of testing to be carried out on these systems. To assess the effect of the elevated metals (aluminum and copper) in the soils used in construction on CWTS functionality, four soil analytical test methods were used:

- Total concentration of elements in the soils by CRC ICPMS (EPA 200.2/6020A). This method uses a heated strong acid digestion with HNO_3 and HCl and is intended to liberate metals that may be environmentally available.

- In 2015 and 2017, leachable concentrations of elements in the soils by Synthetic Precipitation Leaching Procedure (SPLP), which is a method to assess the mobility of elements in soils at the pH of rain water (i.e., if the CWTS were to entirely dry out, then be subjected to leaching by rain water) (EPA 1312/6010B).
- In 2016 and 2017 a leach method was used (EPA 6020A). This analysis was carried out using a leaching procedure which involved the gentle tumbling of a sample in a specified leaching solution (water from the CWTS) for two hours. This method is typically carried out by tumbling the sample in deionized water, however this method was adjusted to use water from the CWTS to be more representative of the actual leachability of the soils into the overlaying water of the CWTS. The resulting extract is then analyzed by inductively coupled plasma – optical emission spectrophotometry (EPA 300.1). This method was selected in 2016 to determine the concentration of metals that are being released from the soil into the overlying water column. The leach method is more representative of actual potential leachability of the soils into the CWTS than is the SPLP method.
- Sequential extraction procedure for the speciation of particulate trace metals (Tessier et al., 1979, EPA 6020A) to assess the stability and form of elements in soils.

A3.4. Plant Sampling and Analyses

Both *C. aquatilis* and moss samples were collected for metals analysis at the final sampling period of the demonstration-scale CWTS. Above water vegetation for *C. aquatilis* samples were collected from each cell in the CWTS. Green, living moss samples (rather than black, decomposing moss) were collected from each cell in the CWTS. Analyses are outlined in Table A5. Analyses of the tissue samples followed EPA 200.3/6020A and were reported in mg/kg dry weight. The tissue samples were homogenized and sub-sampled prior to digestion.

Table A5 - Summary of 2017 analytical plant sampling types, frequencies and locations.

Plant tissue samples	
Total Metals by ICPMS (Co, Cu, Fe, Pb, Li, Mg, Mn, Mo, Ni, P, K, Se, Ag, Na, Sr, Tl, Sn, Ti, U, V, Zn, Zr)	*Carex aquatilis* and aquatic moss, each cell, year end

A3.5. Microbial Sampling

Soil, roots, detritus, and moss were collected from each cell in the CWTS for microbial analysis three times in 2017 (June, July, Sept). The soil was collected by scooping the surface, from 0-10 cm, in various parts of the cell and pouring off any excess water after collection. The roots were collected from various plants in each cell at different depths of soil and pooled together in a 50mL falcon tube. Detritus samples were collected from various areas of the cells and included the decomposing straw material within the water. Moss samples were collected from various parts of the cell from the green new growth as well as the decomposing black areas of the moss and pooled together in a 50mL falcon tube. Microbial analyses are outlined in Section A3.6.

A3.6. Methods for Microbial Analyses

A3.6.1. Growth-based Analyses (MPN)

The most-probable number (MPN) of bacteria was determined for all microbial samples. The MPN test allows for an estimation of the number of bacteria that can grow in a specific laboratory medium (i.e., quantification method). MPN of heterotrophic organisms (grown with R2A medium; HiMedia Labs) were quantified in both aerobic and anaerobic conditions. MPN tests for selenate and selenite reduction were performed as per Siddique et al. (2006). The nitrate reduction MPN was performed as per the Nitrate Reduction Test (supplied by Sigma-Aldrich).

In brief, samples were weighed and/or measured into a 0.1% peptone solution and then serially diluted along a sterile 96-microwell round-bottom plate containing the growth media. All tests were conducted in triplicate. Wells were incubated without light at room temperature (21-22°C) and assessed for visible growth (formation of a pellet) and/or colour change specific to the type of media after 27-29 days. A colour change to red indicated selenite or selenate reduction, while nitrate reduction was assessed per the manufacturer's instructions. The most probable number of microbes was then calculated as described by Blodgett (2010).

A3.6.1. DNA-based Analyses

DNA was extracted from all samples using the MO BIO PowerLyzer PowerSoil DNA Isolation kit. Targeted DNA sequencing was used to identify bacteria present in each sample via polymerase chain reaction (PCR) amplification of the v3/v4 region of the 16S ribosomal RNA gene (Klindworth et al., 2013). Library preparation and sequencing was performed as per the manufacturer's instructions for MiSeq v3 paired-end 300 bp sequencing (Illumina) for all samples and positive and negative controls. After sequencing, the forward and reverse reads were merged and all sequences were quality filtered before processing into Operational Taxonomic Units (OTUs) and classified.

A3.7. Hydraulic Retention Time Tracer Study

The Minto Demonstration Scale 2016 Update Report (Contango, 2017) provides details on the methods used in the HRT tracer study in 2016. In 2017, a 1 m^3 tank was added to the CWTS area, and another HRT tracer study was conducted. The 1 m^3 tank provided a way to mix a lower concentration salt solution, and dose it into the CWTS over a longer period at a constant flow using a metering pump. This allowed the flow rate to be calculated for the duration of time the salt solution was dosed into the CWTS. To conduct the HRT tracer study over a longer period, the YSI unit was connected to a power source due to a limited battery life. By conducting the tracer study in 2017 as described above only one breakthrough occurred, as opposed to the two breakthroughs that were observed in 2016, providing more accurate results.

The 1 m^3 tank used in the HRT tracer study was filled with 600 L of influent water with 1.26 kg of salt dissolved into the water to create the tracer. The salt solution was mixed thoroughly with a mixing pump in the tank prior to use. The salt solution was dosed into cell 1A at a rate

of 1.8 L/min. The tracer was dosed over a period of approximately four hours, at which time approximately 400 L of salt solution had been dosed into cell 1A. The volume of water and weight of salt was calculated based on the amount required to increase the specific conductance (SPC) of cell 1A to approximately twice its background level (i.e., from 850 µs/cm, to 1,700 µs/cm) (Oakton Instruments, 1997). When using this method in 2016 the SPC was raised by 38% higher than desired, therefore in 2017 38% less salt was used for the tracer. The volume of water used was then selected to ensure the concentration of the salt solution was sufficiently diluted and would mix with the water of the CWTS (i.e., the solution wasn't so dense that it would stratify and sink to the bottom of the CWTS).

In situ monitoring was carried out from June 21, 2017 until July 2, 2017. This time period was selected to allow sufficient time for the entire salt tracer to pass through the CWTS and return to background SPC levels. The YSI unit was placed in the CWTS for approximately two hours prior to dosing of the tracer to record background readings. The background SPC prior to commencing the HRT tracer study was approximately 829 µS/cm, and the SPC peaked at 1107 µS/cm, 2.25 days after the tracer was dosed into the CWTS (Figure 20, in main document). The increase in conductivity did not reach the desired amount of double the background SPC level, however the peak was sufficient to observe the tracer pass through the system and determine the maximum.

A3.8. Detritus Decomposition Trials

A detritus decomposition trial was developed and conducted in 2017 to assess decomposition rates of *C. aquatilis* in the CWTS over time, as well as the algae growth on CWTS organic materials within the water column (Chimney and Pietro, 2006; Hammerly et al.,1989). Twenty-four 12-inch by 4-inch mesh bags were filled with 5 grams of oven dried *C. aquatilis* material (dried at 50°C). On June 21, 2017, six bags were then submerged into each of the four CWTS cells to determine the rate of decomposition of *C. aquatilis* over time. An additional 24 mesh bags were filled with 3.8 grams of polyester filter fiber material. On the same day (June 21, 2017) six bags were submerged into each of the four CWTS cells to determine the amount of algae growth on material in the CWTS during the course of the trial. On July 25, 2017, after 34 days of submersion in the CWTS, and on September 11, 2017 after 82 days of submersion in the CWTS, one bag filled with polyester filter fiber and one bag filled with *C. aquatilis* was removed from each CWTS cell. The contents of the bags were then rinsed with deionized water to remove any sediment from the sample and dried in an oven at 50^{o}C. The dried weights after 34 days and 82 days were compared to the initial weights of the sample. It was assumed that the increase in weight on the polyester fiber fill sample would be representative of the algae that grew on the *C. aquatilis* samples that could not be removed by rinsing the samples. Therefore, the increase in dry weight of the polyester fiber was subtracted from the dry weight of the *C. aquatilis* to determine the amount of decomposition that occurred over a specific time period.

In total, six bags of each material were submerged in each cell and two bags of each material were removed over the 2017 sampling periods. The remaining four bags of each material were left in each cell for future analysis. Each removal of four bags (one from each cell) were treated as replicates for the sample material and date removed.

A3.9. Evapotranspiration Trials

Total evapotranspiration from the system is the combined effects of open water evaporation and plant transpiration (Beebe et al, 2014). The purpose of calculating the evapotranspiration of a system is to understand the amount of water lost, which in turn concentrates elements and should be considered in the context of the difference of decrease in outflow concentration (or not) and outflow load reduction. In 2016, an off-site and on-site evapotranspiration study was conducted (Contango, 2017); however, rain occurred during the on-site study and the Minto weather station was not functioning, therefore evapotranspiration rates could not be determined and it was recommended the study be repeated in 2017.

To add to the information gathered from the off-site evapotranspiration trials conducted in 2016 (Contango, 2017), three trials were conducted on-site to refine site-specific evapotranspiration rates. In 2017, the evapotranspiration studies were performed on-site from May 16-24 (trial 1), June 8-11 (trial 2), and June 11-15, 2017 (trial 3). Approximately 17.5 mm of rain occurred on-site during trial 2 on June 11, 2017 and therefore trial 2 was terminated and trial 3 began to account for the precipitation. In trial 1, the flow to the CWTS was shut off on May 16 and the water level was recorded in each cell of each series using depth measuring sticks, which were installed at each end of the CWTS in 2016 (Figure A5). The water level was recorded again on May 24 to determine the water lost due to evapotranspiration in the CWTS throughout the study. The same method to determine water lost was also used in trials 2 and trial 3 and the flow to the CWTS was shut off on June 8 to June 11 and June 11 to June 15, respectively.



Figure A5 – Depth sticks in CWTS cell 1A to record evapotranspiration in 2016 (left) and 2017 (right).
Depth sticks were installed in all CWTS cells June 2016 and used throughout the demonstration-scale study.

A4. Thermodynamic Minimum Calculations

The thermodynamic minimum is the lowest concentration consistently achievable for a given treatment design and water chemistry. Once reached, making the CWTS bigger will not result in further decrease of concentration (although it will continue to decrease the load). Therefore, this is a useful parameter to aid in determining appropriate sizing and outflow water quality objectives of a CWTS. The CWTS was analyzed at three points to determine whether the thermodynamic minimum had been reached. Concentrations of constituents were measured at three sampling points (CWTS feed, cell A-outflow, and cell B-outflow) in the series and compared to determine if the thermodynamic minimum had been reached. To determine if a

concentration was significantly different or not, statistical analyses were done (paired samples t-test, α=0.05) to determine if the concentration had stabilized across two sampling points. Once a concentration had stabilized across two points in a series, the thermodynamic minimum was considered to have been met.

A5. Load Removal Calculations

The load removed shows how much of the load of a constituent entering the pilot-scale CWTS was removed during each flow period. Equation A1, Equation A2, and Equation A3 show how the nominal load in, load out, and load removal (non-evapotranspiration adjusted) is calculated.

$$Load_{in} = C_i \times V$$

Equation A1 – Equation for the calculation of load of a constituent into a CWTS over a period of time.

$Load_{in}$ is the mass of a constituent that enters a CWTS over a period of time, C_i is the inflow concentration of the constituent, V is the volume of water that enters a CWTS over a period of time.

$$Load_{out} = C_f \times V$$

Equation A2 – Equation for the calculation of the load of a constituent out of a CWTS over a period of time.

$Load_{out}$ is the mass of a constituent that exits a CWTS over a period of time, C_f is the outflow concentration of the constituent, V is the volume of water that enters a CWTS over a period of time.

$$Load\ Removed = Load_{in} - Load_{out}$$

Equation A3 – Equation for the calculation of the load removed over a period of time.

A5.1. Load Removal Calculations Adjusted for Evapotranspiration

The results of the evapotranspiration trial (Section A3.9) were integrated with the load removal calculations (Section A4) to determine the actual amount of load removed taking into consideration the amount of water lost from the pilot-scale CWTS through evapotranspiration (Equation A4). Equation A5 describes how the outflow water volume from a CWTS is adjusted for evapotranspiration. Equation A5 uses a value for evapotranspiration in L/day/m^2, which was calculated using the results of the evapotranspiration trial (main report), which is multiplied by the surface area of the CWTS. For series 1 the surface area is 69.7 m^2, and for series 2 the surface area is 58.2 m^2. Equation A6 describes how the load out adjusted for evapotranspiration is calculated using the evapotranspiration adjusted outflow volume calculated in Equation A5. Equation A7 was used to calculate the outflow concentration (mg/L) adjusted for evapotranspiration using the load out calculated in Equation A6. Equation A8 describes how the load removal adjusted for evapotranspiration is calculated using the load in calculated in Equation A1 and load out adjusted for evapotranspiration in Equation A6.

$$Water\ loss\ (\%) = \frac{ET}{Q} \times 100$$

Equation A4 – Equation for the calculation of the water loss due to evapotranspiration.
Q is the flow rate (L/day), *ET* is the evapotranspiration rate (L/day).

$$V_{ET} = Q - ET$$

Equation A5 – Equation for the calculation of the outflow water volume adjusted for evapotranspiration.
V_{ET} is the outflow water volume adjusted for evapotranspiration, *Q* is the flow rate (L/day), *ET* is the evapotranspiration rate (L/day).

$$Load_{out(ET)} = V_{ET} * C_f$$

Equation A6 – Equation for the calculation of the load out of a constituent adjusted for evapotranspiration.
$Load_{out(ET)}$ is the load out adjusted for evapotranspiration, V_{ET} is the outflow water volume adjusted for evapotranspiration, C_f is the outflow concentration of the constituent.

$$Cf_{ET} = \frac{Load_{out(ET)}}{Q}$$

Equation A7 - Equation for the calculation of outflow concentration adjusted for evapotranspiration.
Cf_{ET} is the outflow concentration of the constituent adjusted for evapotranspiration, $Load_{out(ET)}$ is the load out adjusted for evapotranspiration, *Q* is the flow rate (L/day).

$$Load\ Removed\ Adjusted\ for\ Evapotranspiration = Load_{in} - Load_{out(ET)}$$

Equation A8 – Equation for the calculation of the load removed adjusted for evapotranspiration.
$Load_{out(ET)}$ is the load out adjusted for evapotranspiration

A6. Removal Rate Coefficients and Calculations

An important factor for CWTS design is the rate of treatment, also known as the removal rate coefficient (*k*). The removal rate coefficient is based on the treatability of a specific compound and the hydraulic retention time of the CWTS, both of which are site-specific based on water chemistry, CWTS designs, and characteristics of the CWTS.

Once hydraulic retention time is calculated (main report), it can be used to solve for the removal rate coefficients (*k*) using the inflow (*Ci*) and outflow (*Cf*) concentrations of a given constituent (Equation A9 and Equation A10). Equation A9 was utilized to calculate first-order removal rate coefficients, while Equation A10 was utilized to calculate zero-order removal rate coefficients.

$$k = \frac{-\ln\left(\frac{Cf}{Ci}\right)}{\mathrm{HRT}}$$

Equation A9 – Equation for calculation of first-order removal rate coefficient.
k is the removal rate coefficient, *Cf* is the final concentration, *Ci* is the initial concentration, and HRT is the hydraulic retention time.

$$k = \frac{(Ci - Cf)}{\mathrm{HRT}}$$

Equation A10 – Equation for calculation of zero-order removal rate coefficient.
k is the removal rate coefficient, *Cf* is the final concentration, *Ci* is the initial concentration, and HRT is the hydraulic retention time.

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Table B1 - Summary of extractable fractions from sequential extraction procedure for the speciation of metals[1].

Fraction	Description	COCs unstable when
1	Exchangeable fraction for adsorbed minerals	Readily released (i.e., soluble and exchangeable)
2	Mineral fraction bound to carbonates or solubilized at pH 5	Decreased pH
3	Oxidized mineral fraction bound to Fe-Mn oxides	Reducing conditions
4	Reduced mineral fraction and sulphides	Oxidizing conditions
5	Residual mineral fraction (primary and secondary minerals)	Not expected to be released in solution over time under conditions normally encountered in nature

[1]Method based on Tessier et al., 1979.

Table B2 – Total and leachable soil copper concentrations in first year of operations.

CWTS Cell	Sampling Date	Days in Operation	Sample Depth (cm)	Total Cu (mg/kg)	SPLP Cu (mg/L)	Leachable Cu (mg/kg)
1A	28-Aug-14	1	Initial Composite	-	0.298 (0.148-0.608)	-
	19-Jun-15	58	0-5	960	0.055[1]	-
	16-Aug-15	116	10-20	950	0.187	-
	18-Sep-15	149	10-20	1300	0.049	-
	29-Sep-15	160	10-20	910	0.069	-
	15-Jun-16	194	0-10	1440	-	0.953
	15-Jun-16	194	10-20	1210	-	0.66
	8-Jul-16	217	0-10	1430	-	0.603
	8-Jul-16	217	10-20	1730	-	0.832
	7-Sep-16	247	0-10	1290	-	0.449
	20-Jun-17	336	0-10	1470	0.315	0.308
	25-Jul-17	371	0-10	1630	0.529	0.463
	12-Sep-17	420	0-10	1490	0.590	0.481
1B	28-Aug-14	1	Initial Composite	-	0.298 (0.148-0.608)	-
	19-Jun-15	58	0-5	1400	0.033[1]	-
	16-Aug-15	116	10-20	1400	0.209	-
	18-Sep-15	149	10-20	830	0.065	-
	29-Sep-15	160	10-20	880	0.059	-
	15-Jun-16	194	0-10	1130	-	1.01
	15-Jun-16	194	10-20	1240	-	0.822
	8-Jul-16	217	0-10	1250	-	1.11
	8-Jul-16	217	10-20	1620	-	1.21
	7-Sep-16	247	0-10	1190	-	0.197
	20-Jun-17	336	0-10	1290	0.450	0.338
	25-Jul-17	371	0-10	1100	0.290	0.165
	12-Sep-17	420	0-10	1100	0.716	0.492

[1] Samples collected in June 2015 were at a shallow depth (0-5 cm) and copper content had therefore likely already been removed by washing from the faster flows of the CWTS system. The blue shading indicated samples that were taken from a shallower depth (0-10).

Table B2 continued – Total and leachable soil copper concentrations in first year of operations.

CWTS Cell	Sampling Date	Days in Operation	Sample Depth (cm)	Total Cu (mg/kg)	SPLP Cu (mg/L)	Leachable Cu (mg/kg)
2A	28-Aug-14	1	Initial Composite	-	0.298 (0.148-0.608)	-
	19-Jun-15	58	0-5	1175	0.037[1]	-
	16-Aug-15	116	10-20	660	0.139	-
	18-Sep-15	149	10-20	880	0.081	-
	29-Sep-15	160	10-20	1000	0.073	-
	15-Jun-16	194	0-10	1100	-	0.823
	15-Jun-16	194	10-20	1280	-	0.900
	8-Jul-16	217	0-10	1450	-	0.963
	8-Jul-16	217	10-20	1150	-	1.68
	7-Sep-16	247	0-10	1290	-	0.838
	20-Jun-17	336	0-10	1820	0.381	1.24
	25-Jul-17	371	0-10	1290	0.419	0.927
	12-Sep-17	420	0-10	1440	0.389	0.381
2B	28-Aug-14	1	Initial Composite	-	0.298 (0.148-0.608)	-
	19-Jun-15	58	0-5	1100	0.039[1]	-
	16-Aug-15	116	10-20	1000	0.201	-
	18-Sep-15	149	10-20	830	0.078	-
	29-Sep-15	160	10-20	540	0.059	-
	15-Jun-16	194	0-10	1040	-	0.771
	15-Jun-16	194	10-20	1310	-	1.16
	8-Jul-16	217	0-10	1070	-	1.38
	8-Jul-16	217	10-20	1180	-	1.25
	7-Sep-16	247	0-10	1520	-	0.123
	20-Jun-17	336	0-10	1110	0.283	0.523
	25-Jul-17	371	0-10	1240	0.477	0.540
	12-Sep-17	420	0-10	1140	0.596	0.809

The blue shading indicated samples that were taken from a shallower depth (0-10).

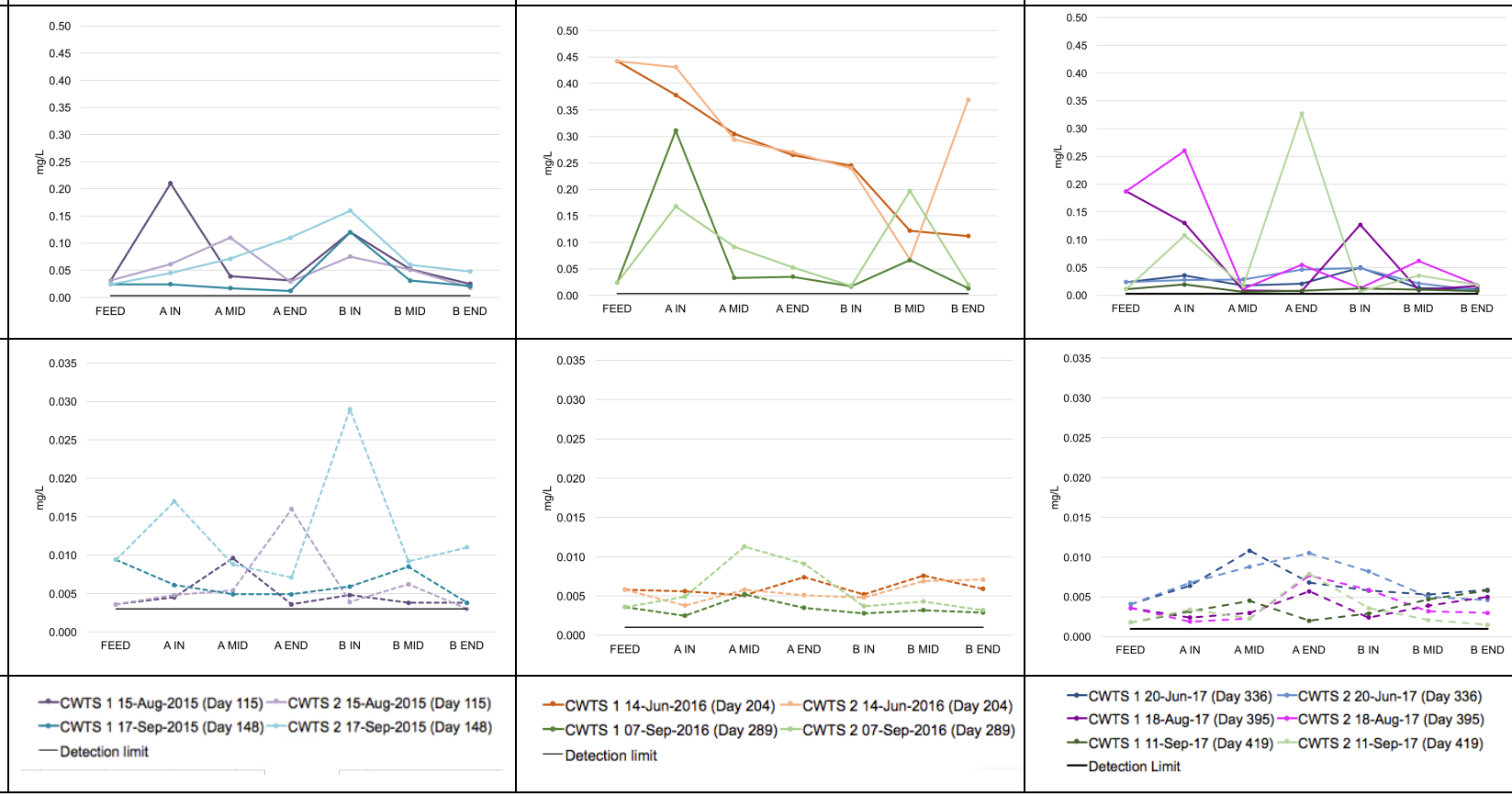


Figure B1 – Aluminum concentrations through the CWTS.
2015 (left), 2016 (middle), and 2017 (right) total (top) and dissolved (bottom) aluminum concentrations. Data shown for seven timepoints, where water was sampled at 7 locations through the flow path of the CWTS to assess for treatment fronts within the wetland, or possible leaching of elements from the soils into the CWTS. The Maxxam (2015 results) detection limit (DL; black line) for aluminum is 0.0030 mg/L. The ALS (2016 and 2017 results) DL for dissolved and total aluminum are 0.0010 mg/L and 0.0030 mg/L, respectively.

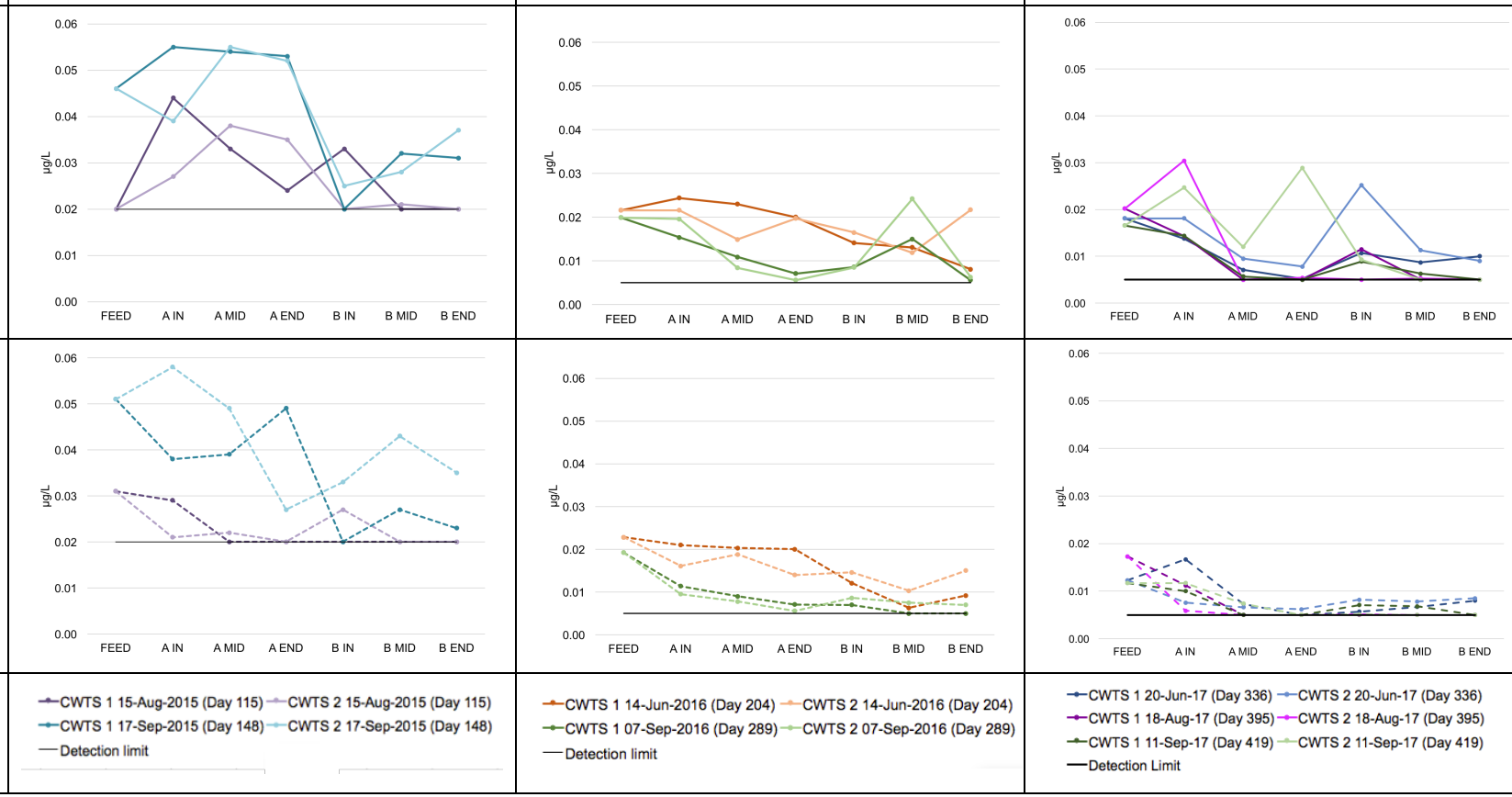


Figure B2 – Cadmium concentrations through the CWTS.
2015 (left), 2016 (middle), and 2017 (right) total (top) and dissolved (bottom) cadmium concentrations. Data shown for seven timepoints, where water was sampled at 7 locations through the flow path of the CWTS to assess for treatment fronts within the wetland, or possible leaching of elements from the soils into the CWTS. The Maxxam (2015 results) detection limit (DL; black line) for cadmium is 0.020 µg/L. The ALS (2016 and 2017 results) DL for cadmium is 0.005 µg/L.

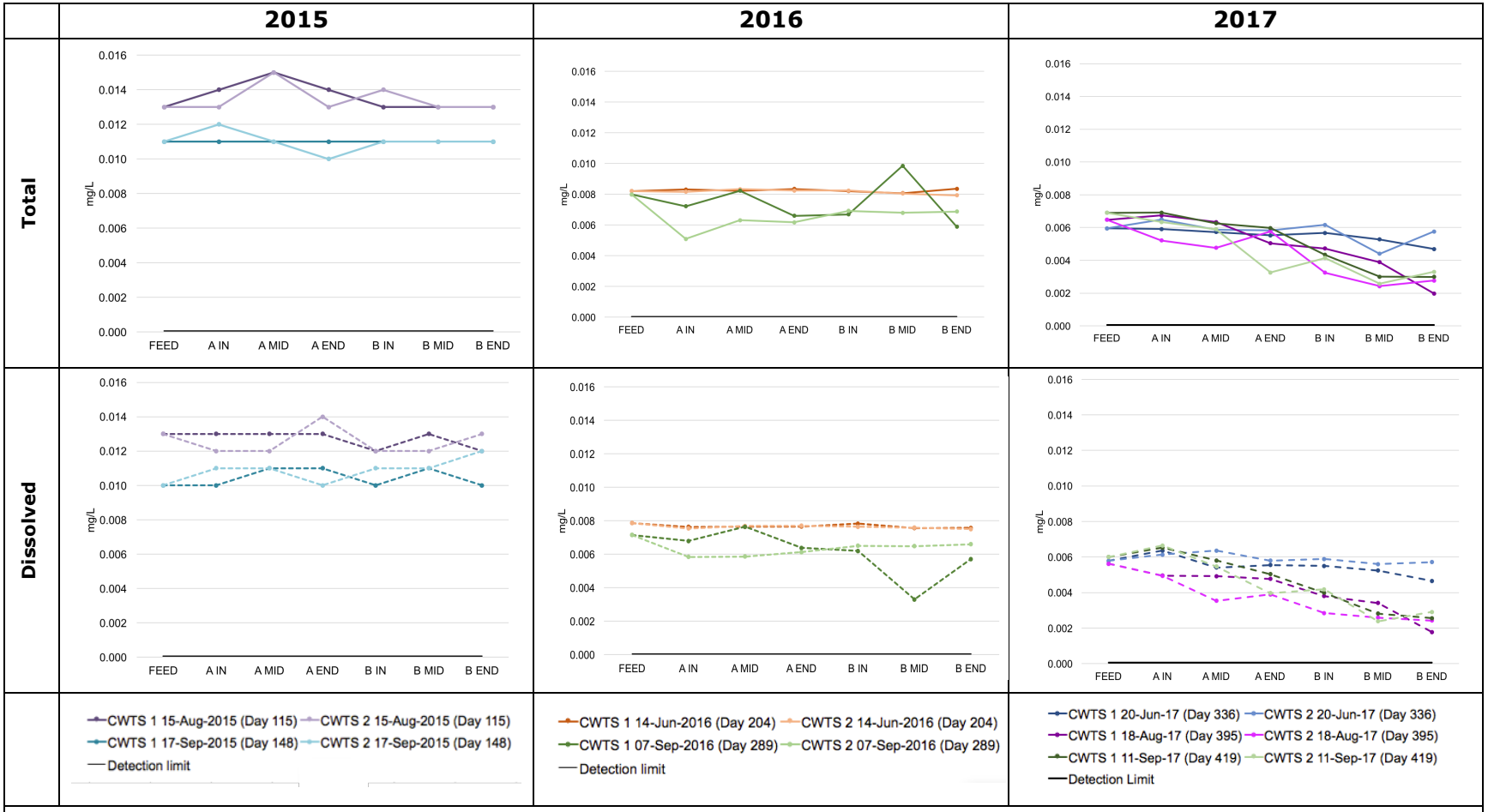


Figure B3 – Molybdenum concentrations through the CWTS.
2015 (left), 2016 (middle), and 2017 (right) total (top) and dissolved (bottom) molybdenum concentrations. Data shown for seven timepoints, where water was sampled at 7 locations through the flow path of the CWTS to assess for treatment fronts within the wetland, or possible leaching of elements from the soils into the CWTS. The Maxxam (2015 results) detection limit (DL; black line) for molybdenum is 0.0002 mg/L. The ALS (2016 and 2017 results) DL for molybdenum is 0.000050 mg/L.

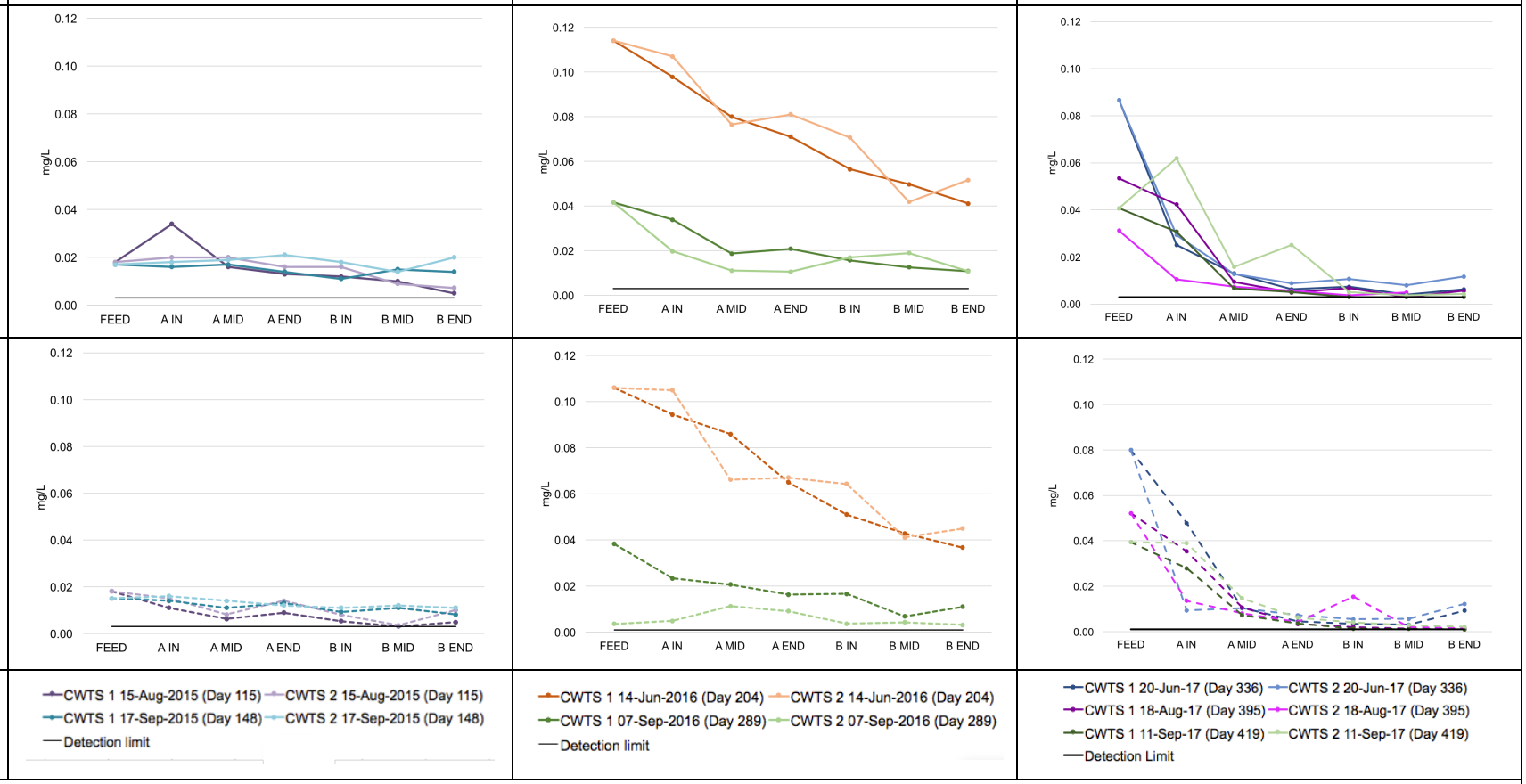
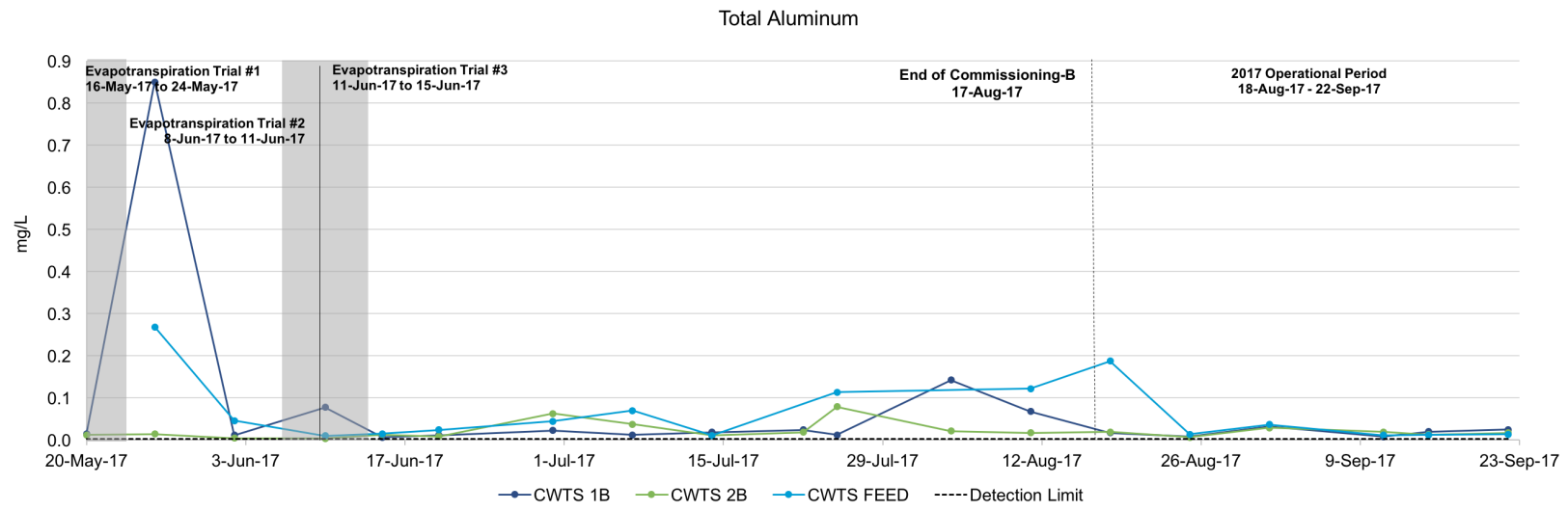


Figure B4 – Zinc concentrations through the CWTS.
2015 (left), 2016 (middle), and 2017 (right) total (top) and dissolved (bottom) zinc concentrations. Data shown for seven timepoints, where water was sampled at 7 locations through the flow path of the CWTS to assess for treatment fronts within the wetland, or possible leaching of elements from the soils into the CWTS. The Maxxam (2015 results) detection limit (DL; black line) for zinc is 0.0030 mg/L. The ALS (2016 and 2017 results) DL zinc is for 0.0010 mg/L.



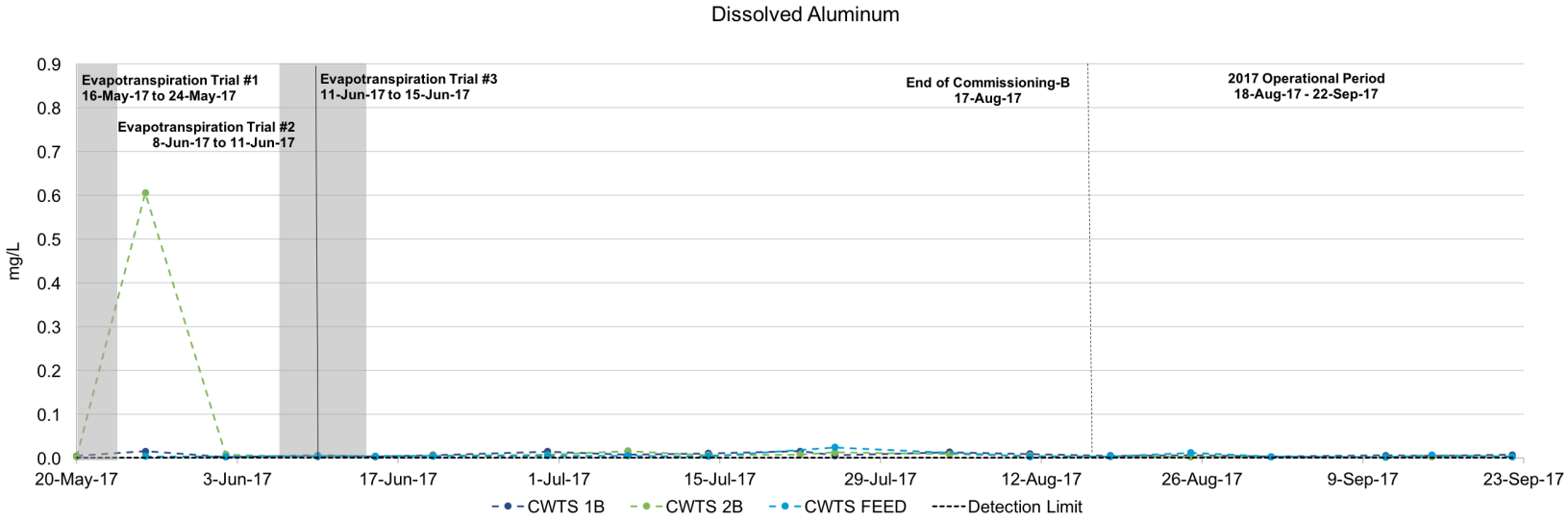
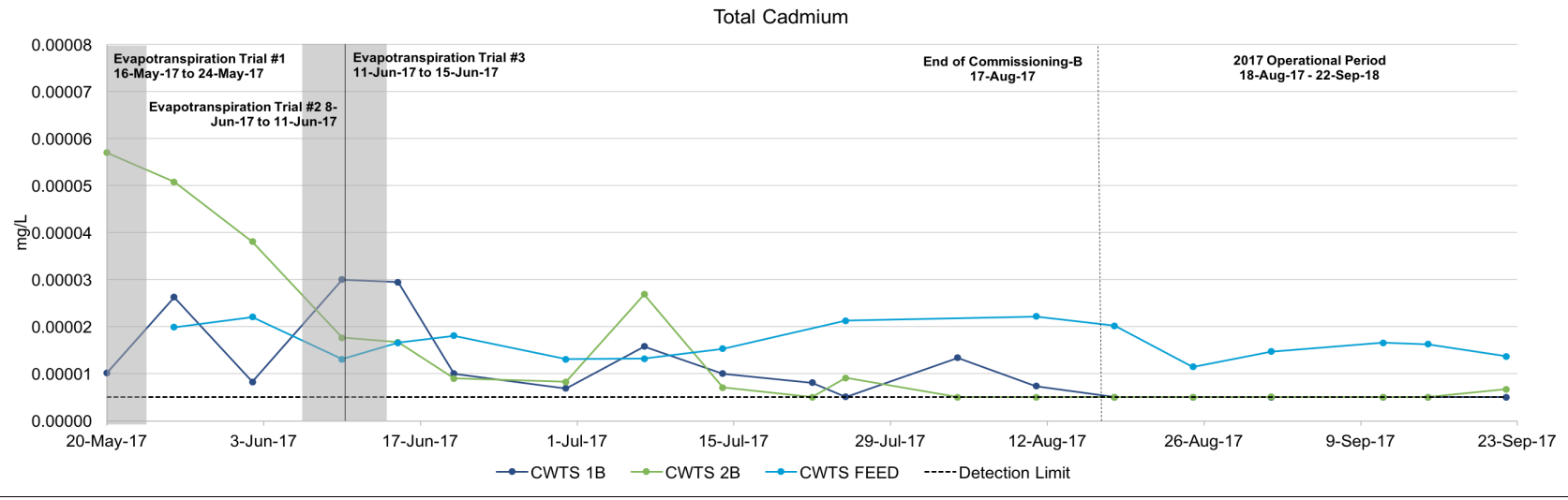


Figure B5 – Aluminum concentrations in the demonstration-scale wetland in 2017.
The evapotranspiration trials are indicated by grey shading and the duration in days. The ALS detection limit for dissolved and total aluminum is 0.001 mg/L and 0.003 mg/L, respectively. Flow to the CWTS was turned off on September 30, 2017. However, the last sampling date in the operational period was September 22, 2017 due to variable flow rates in the CWTS following this date.



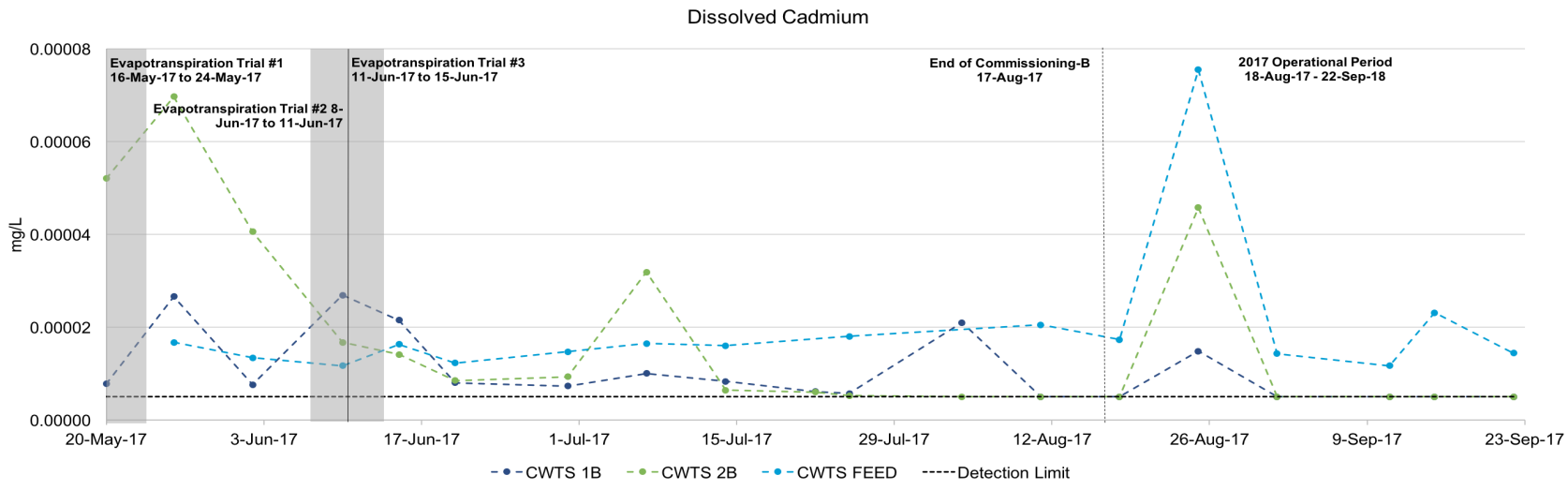
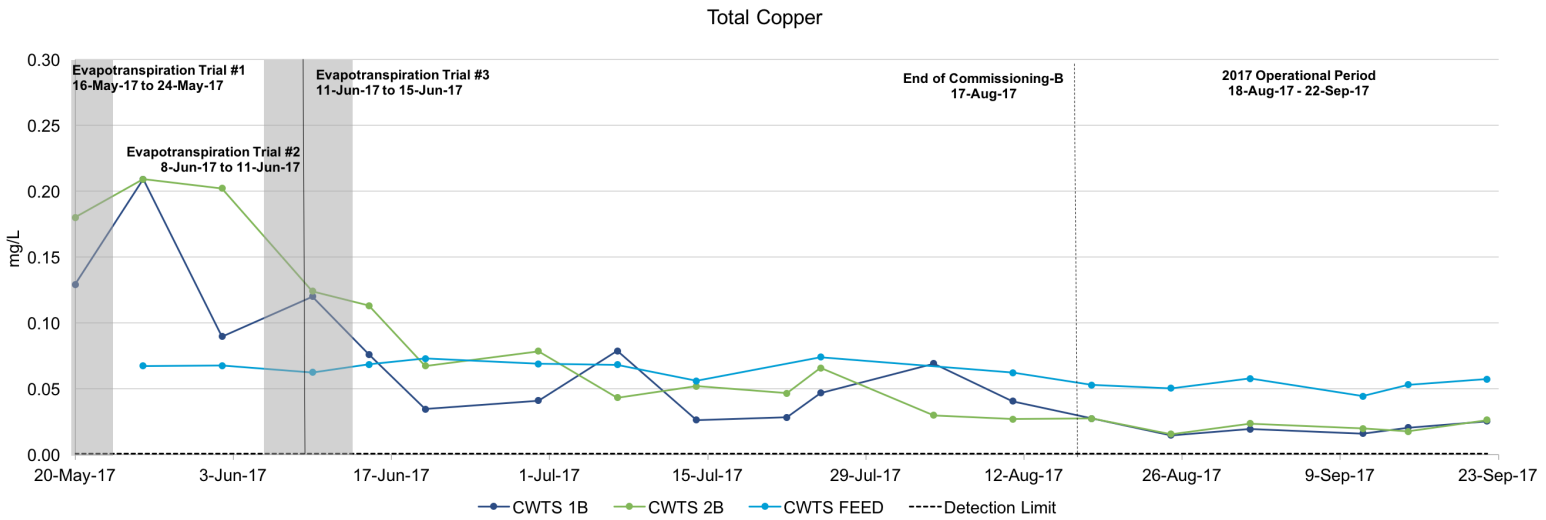


Figure B6 – Cadmium concentrations in the demonstration-scale wetland in 2017.
The evapotranspiration trials are indicated by grey shading and the duration in days. The ALS detection limit for cadmium is 0.000005 mg/L. Flow to the CWTS was turned off on September 30, 2017. However, the last sampling date in the operational period was September 22, 2017 due to variable flow rates in the CWTS following this date.



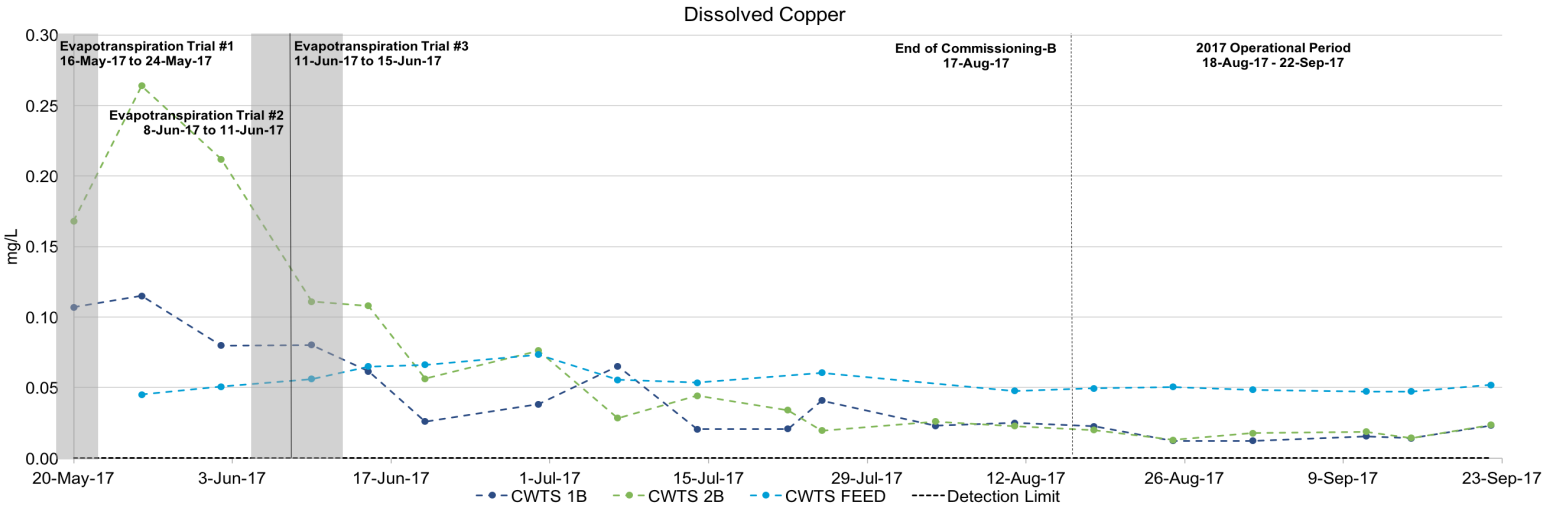
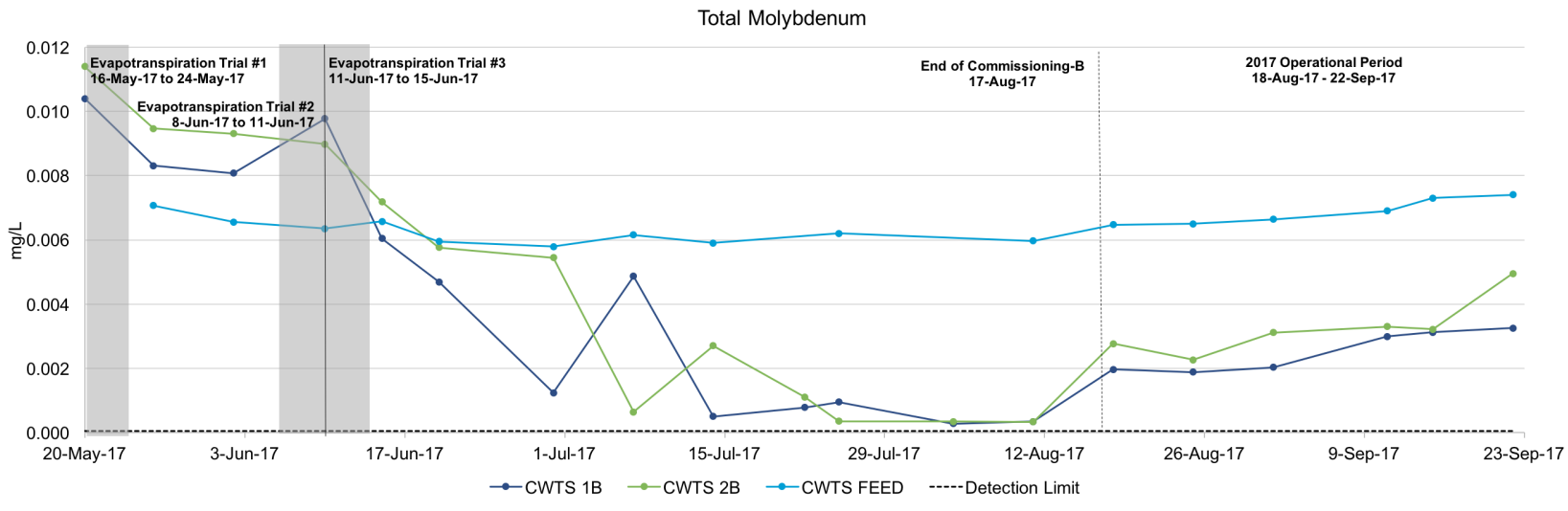


Figure B7 – Copper concentrations in the demonstration-scale wetland in 2017.
The evapotranspiration trials are indicated by grey shading and the duration in days. The ALS detection limit for dissolved and total copper is 0.0002 mg/L and 0.0005 mg/L, respectively. Flow to the CWTS was turned off on September 30, 2017. However, the last sampling date in the operational period was September 22, 2017 due to variable flow rates in the CWTS following this date.



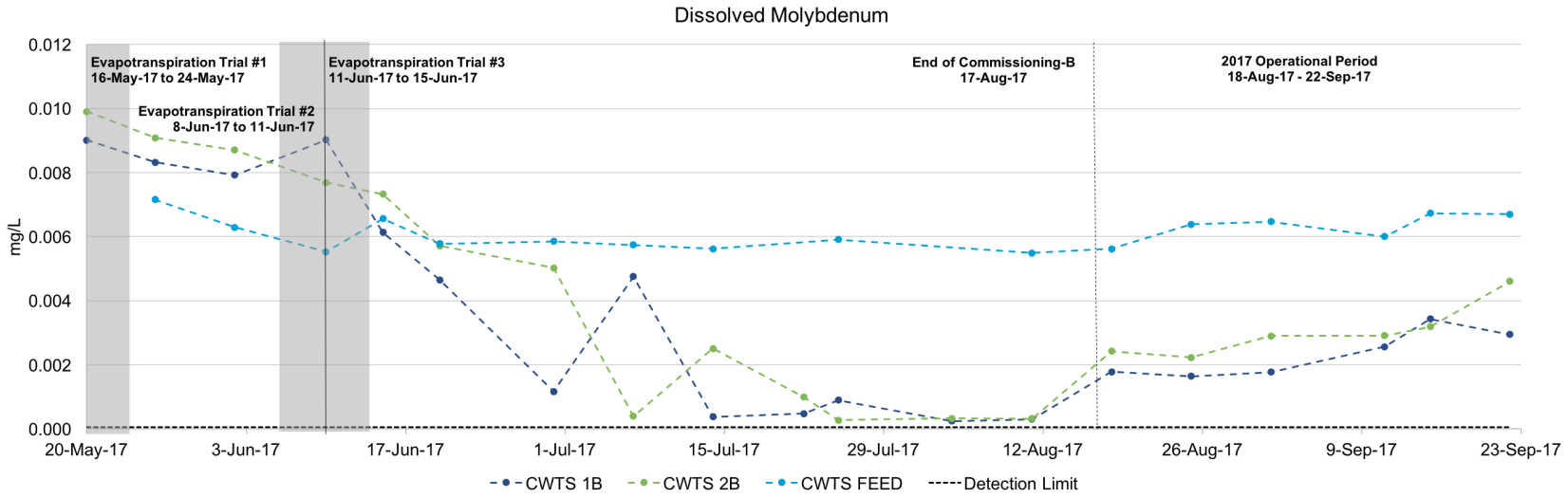


Figure B8 – Molybdenum concentrations in the demonstration-scale wetland in 2017.
The evapotranspiration trials are indicated by grey shading and the duration in days. The ALS detection limit for molybdenum is +/- 0.000050 mg/L. Flow to the CWTS was turned off on September 30, 2017. However, the last sampling date in the operational period was September 22, 2017 due to variable flow rates in the CWTS following this date.

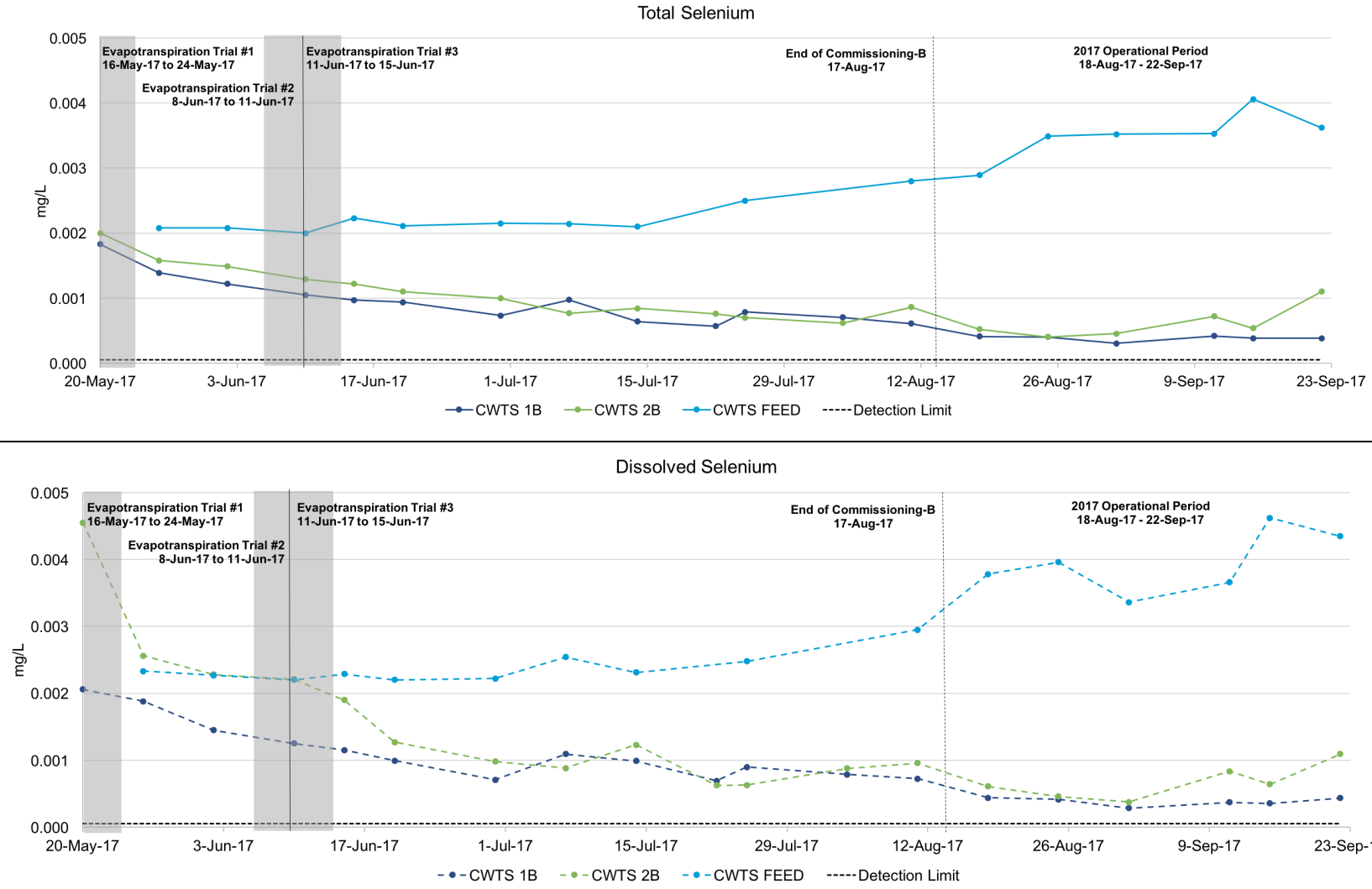


Figure B9 – Selenium concentrations in the demonstration-scale wetland in 2017.
The evapotranspiration trials are indicated by grey shading and the duration in days. The ALS detection limit for selenium is 0.00005 mg/L. Flow to the CWTS was turned off on September 30, 2017. However, the last sampling date in the operational period was September 22, 2017 due to variable flow rates in the CWTS following this date.

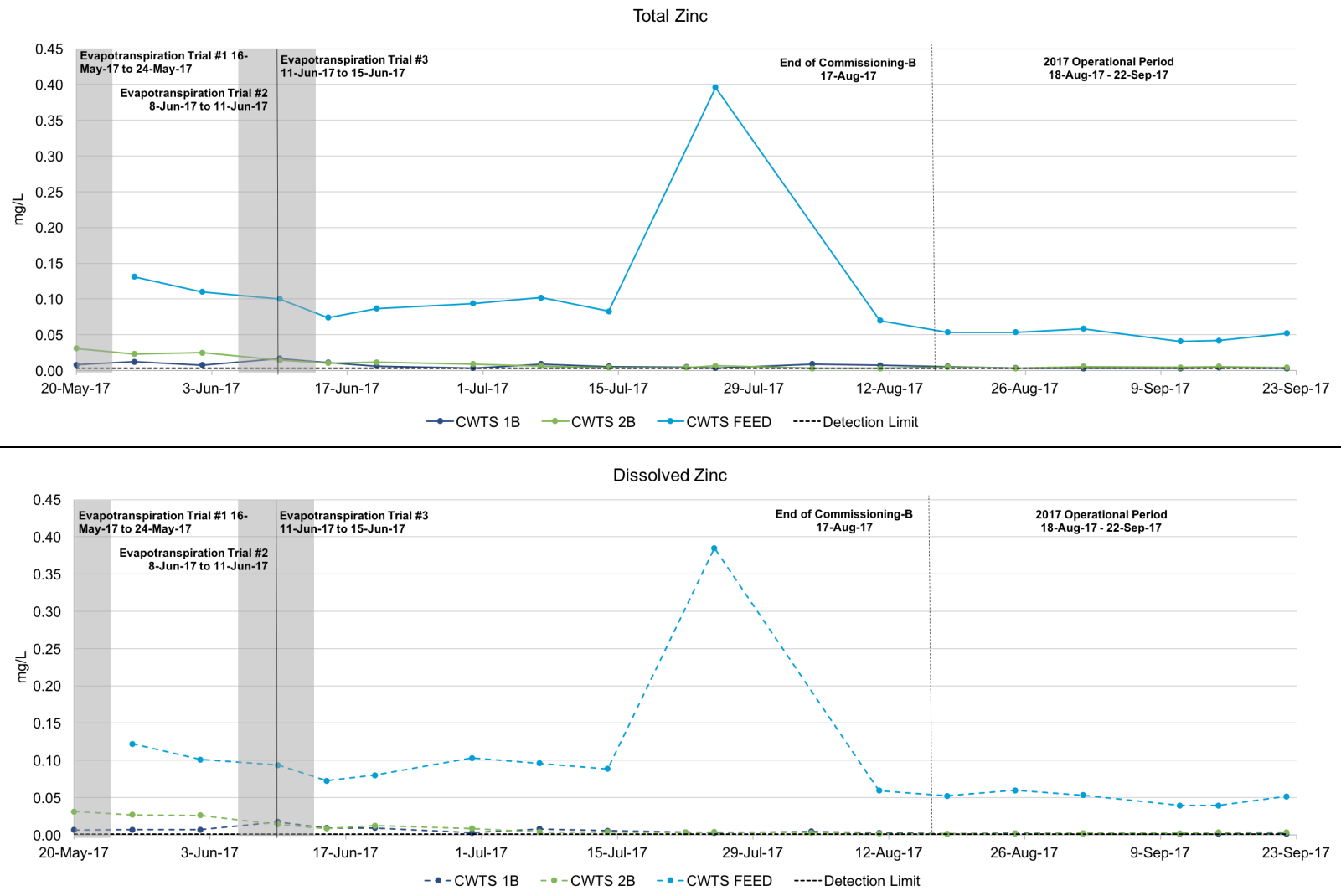


Figure B10 – Zinc concentrations in the demonstration-scale wetland in 2017.
The evapotranspiration trials are indicated by grey shading and the duration in days. The ALS detection limit for dissolved zinc is 0.001 mg/L. The detection limit for total zinc is 0.003 mg/L. Flow to the CWTS was turned off on September 30, 2017. However, the last sampling date in the operational period was September 22, 2017 due to variable flow rates in the CWTS following this date.

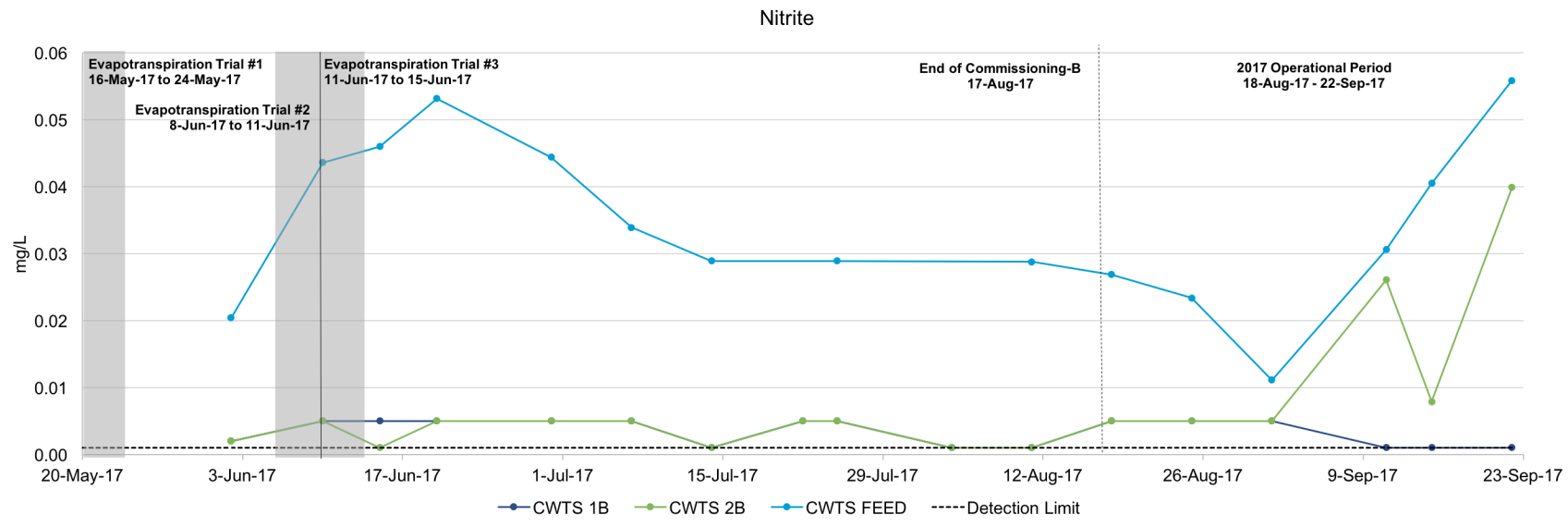


Figure B11 – Nitrite as N (NO_2) concentrations in the demonstration scale wetland in 2017.
The evapotranspiration trials are indicated by grey shading and the duration in days. The ALS detection limit for nitrite is 0.0010 mg/L. Flow to the CWTS was turned off on September 30, 2017. However, the last sampling date in the operational period was September 22, 2017 due to variable flow rates in the CWTS following this date.

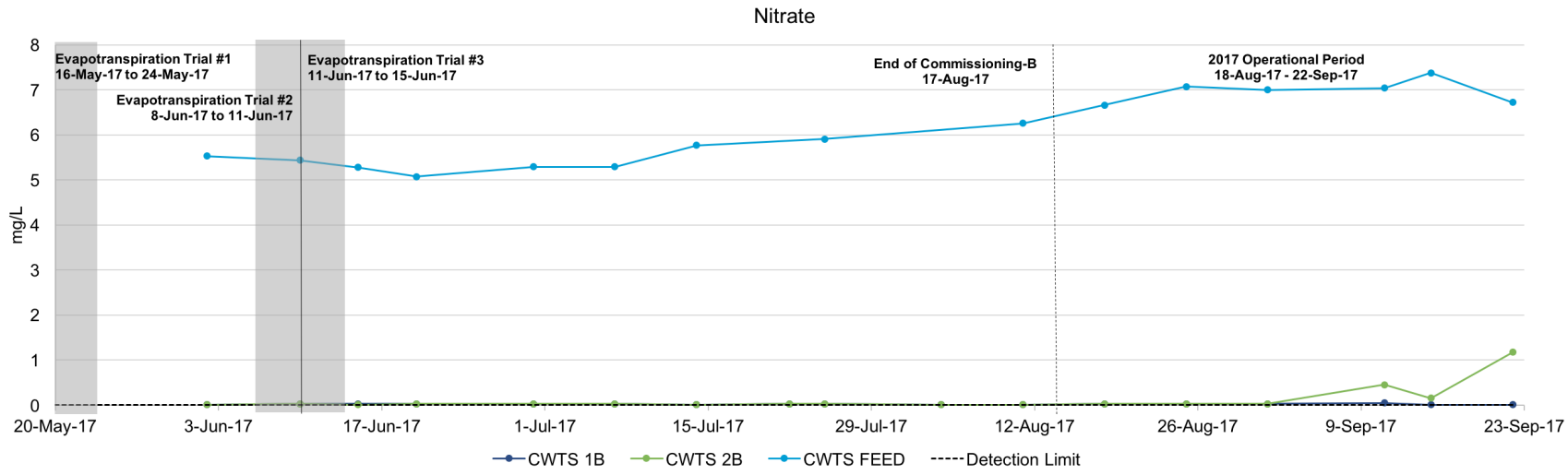
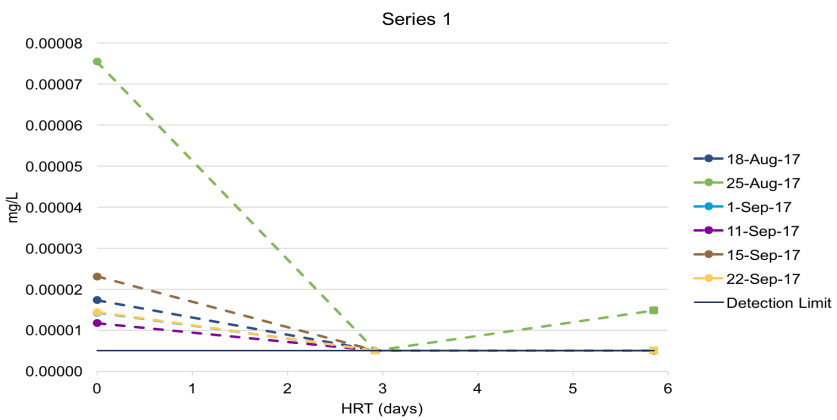
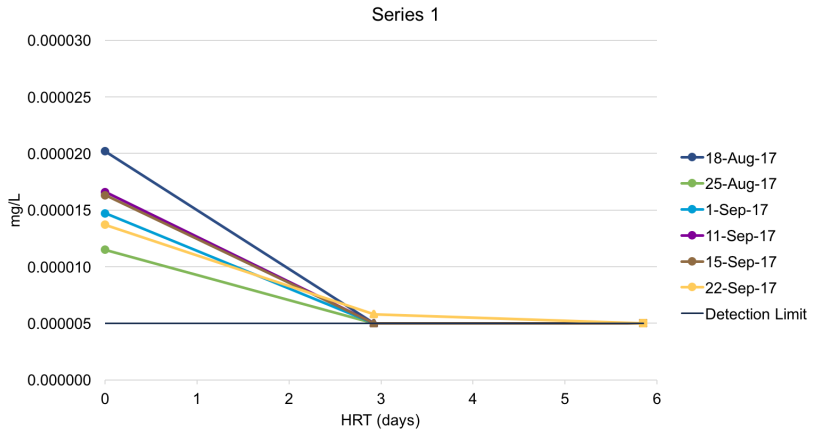
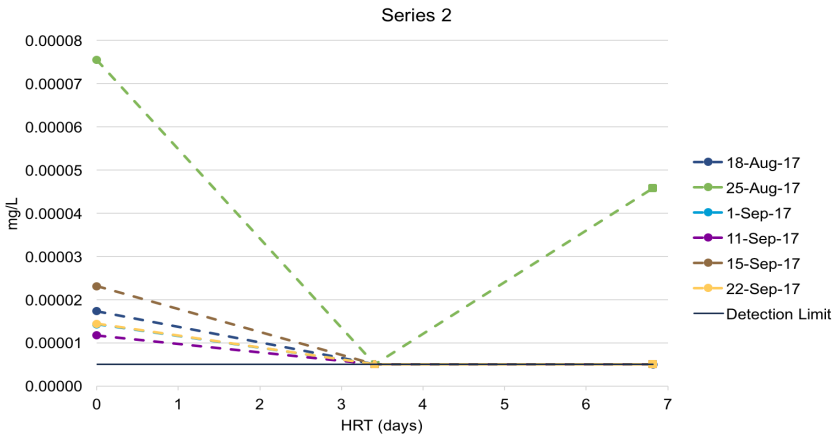


Figure B12 –Nitrate as N (NO_3) concentrations in the demonstration-scale wetland.
The evapotranspiration trials are indicated by grey shading and the duration in days. The ALS detection limit for nitrate is 0.0050 mg/L. Flow to the CWTS was turned off on September 30, 2017. However, the last sampling date in the operational period was September 22, 2017 due to variable flow rates in the CWTS following this date.







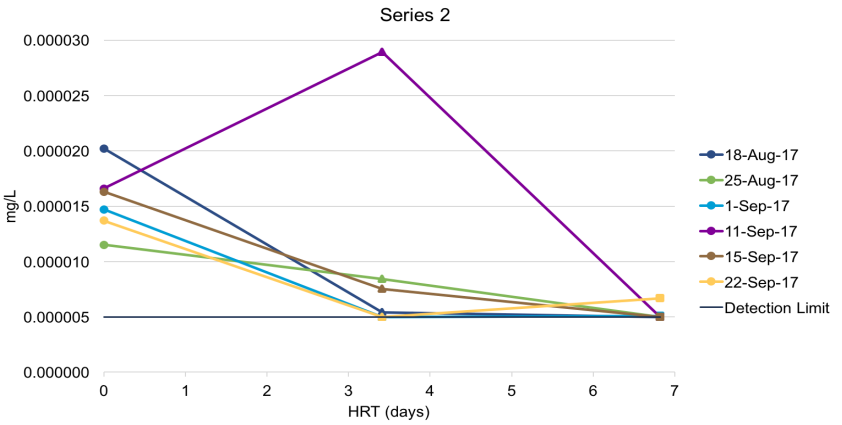
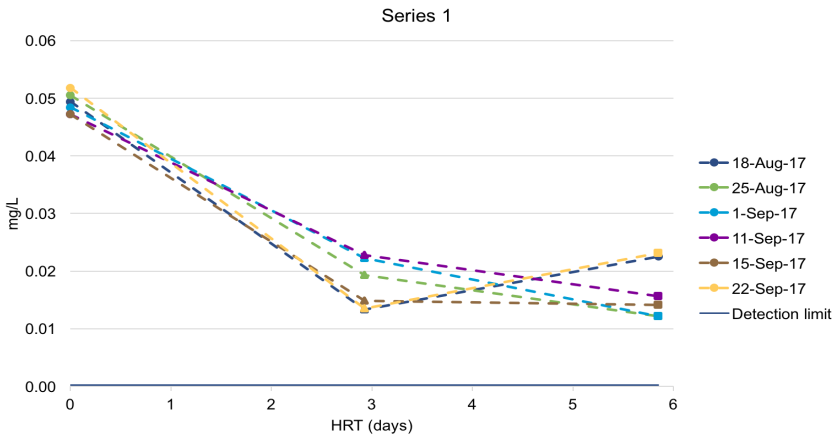
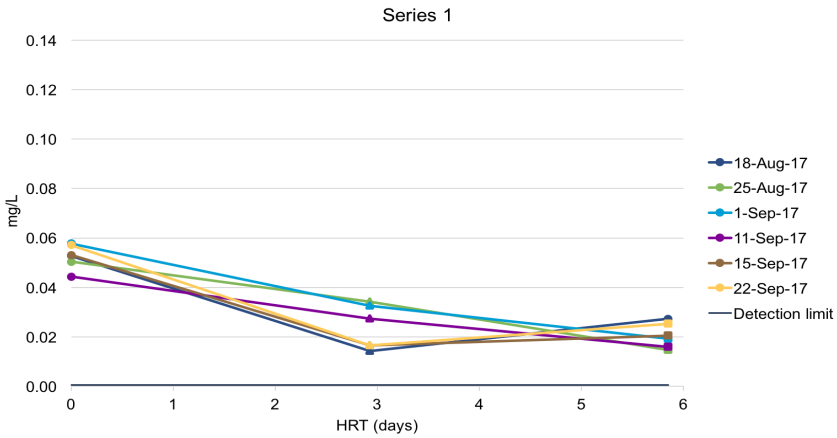
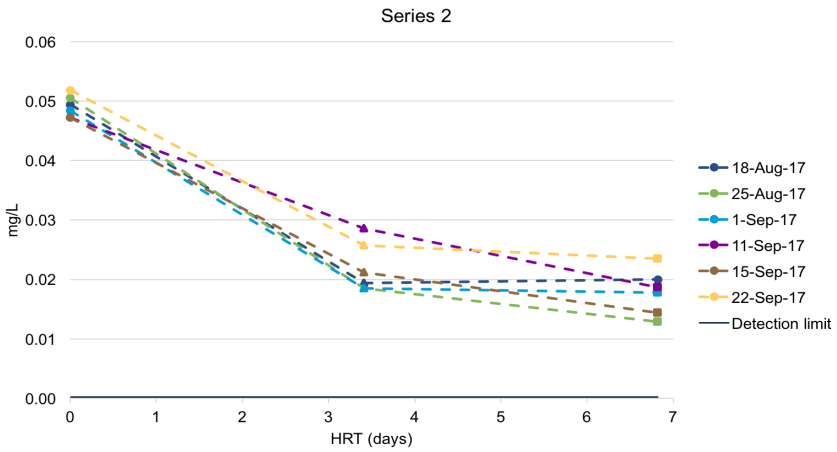


Figure B13 – Dissolved and total cadmium concentrations through series 1 and series 2.
Dashed lines represent dissolved concentrations (left) while solid lines represent total concentrations (right). Circles represent the feed water entering the wetland, diamonds represent the outflow of cell A and squares represent the outflow of cell B. The ALS detection limit for cadmium is 0.000005 mg/L.







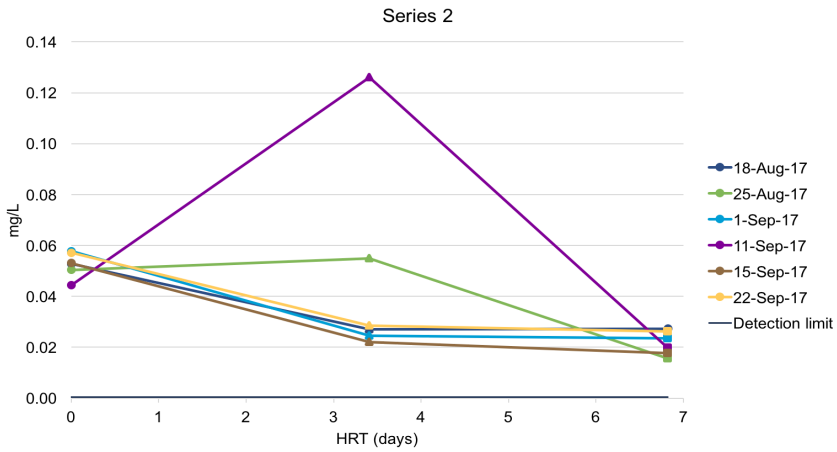
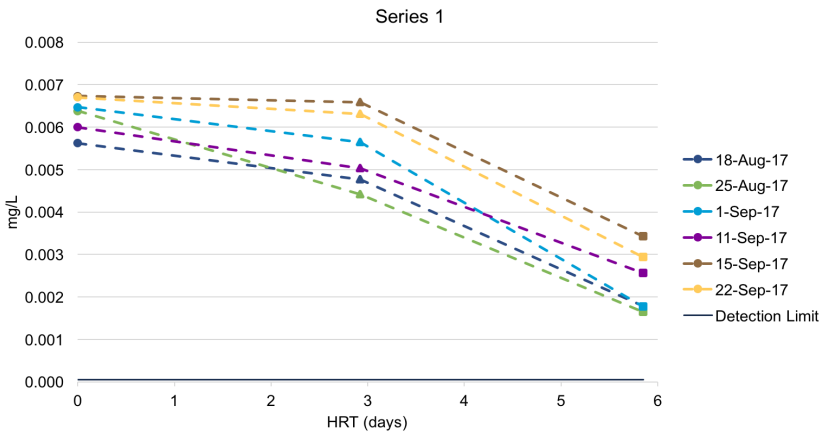
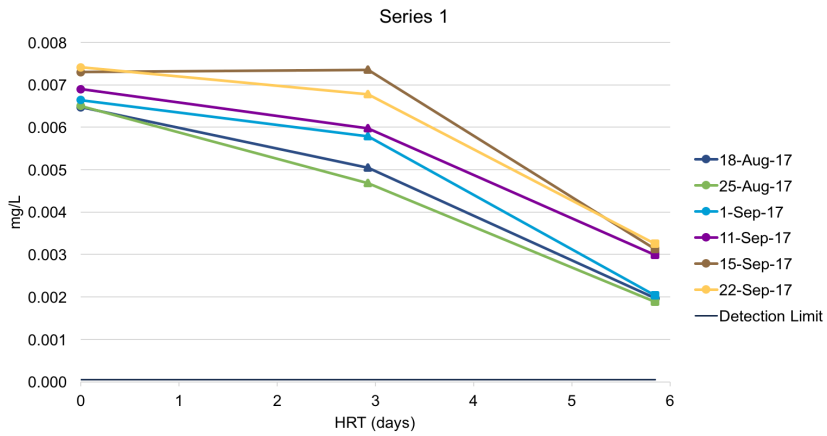
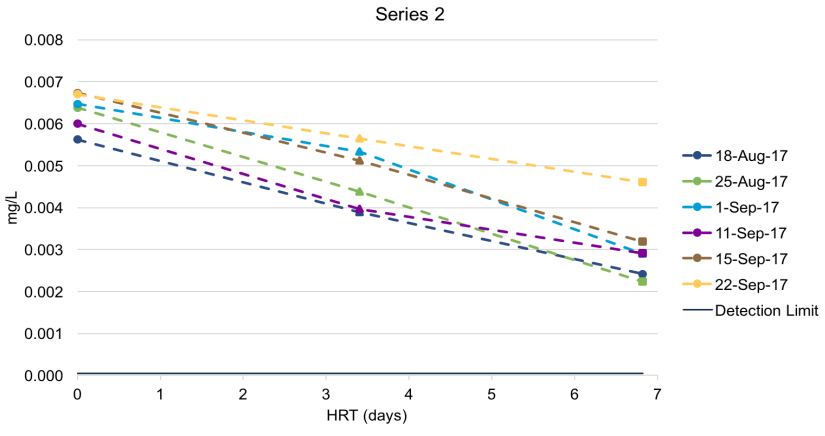


Figure B14 - Dissolved and total copper concentrations through series 1 and series 2.
Dashed lines represent dissolved concentrations (left) while solid lines represent total concentrations (right). Circles represent the feed water entering the wetland, diamonds represent the outflow of cell A and squares represent the outflow of cell B. The ALS detection limit for dissolved copper is 0.0002 mg/L. The detection limit for total copper is 0.0005 mg/L.







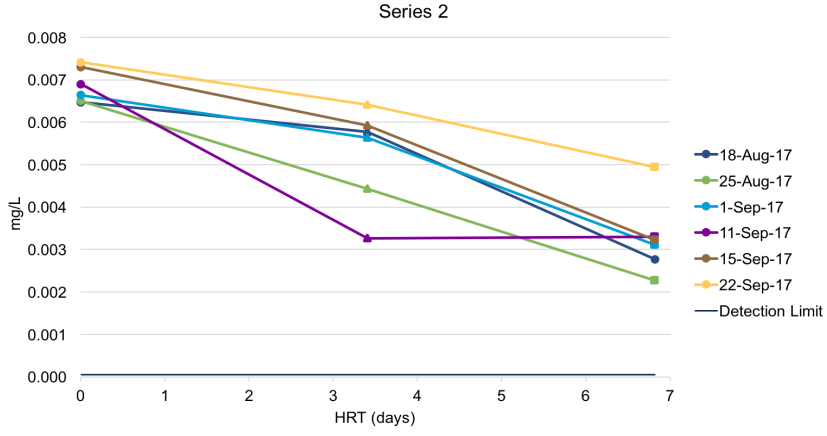
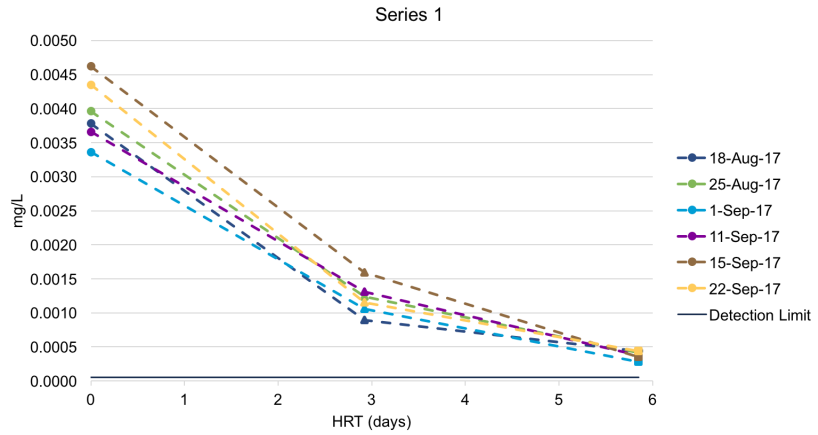
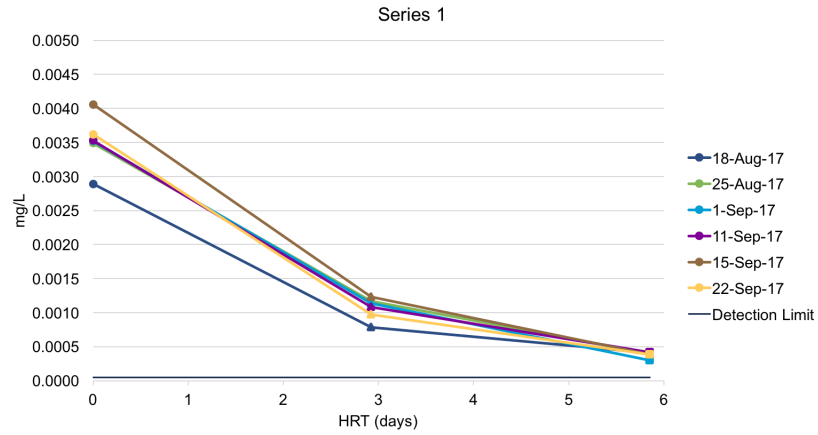
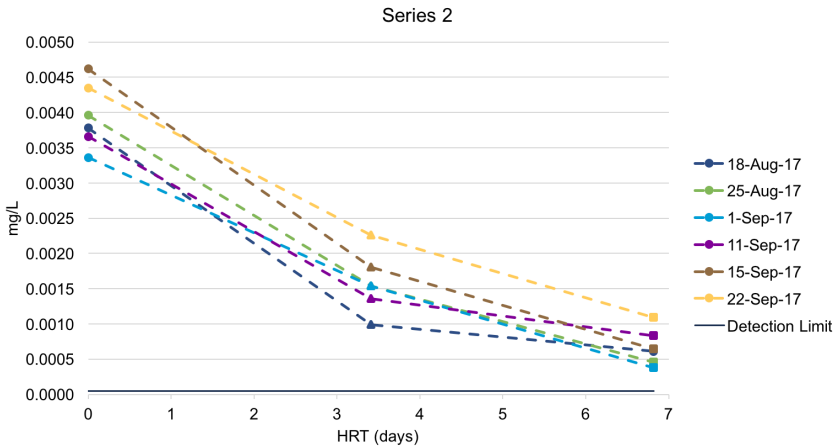


Figure B15 - Dissolved and total molybdenum concentrations through series 1 and series 2.
Dashed lines represent dissolved concentrations (left) while solid lines represent total concentrations (right). Circles represent the feed water entering the wetland, diamonds represent the outflow of cell A and squares represent the outflow of cell B. The ALS detection limit for molybdenum is 0.00005 mg/L.







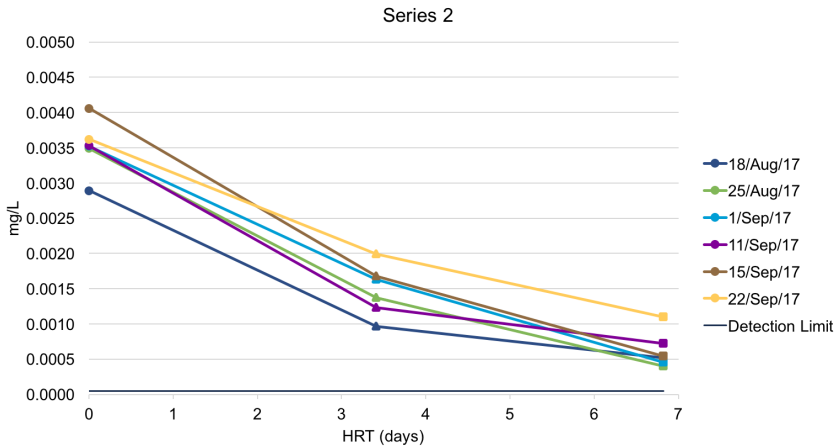
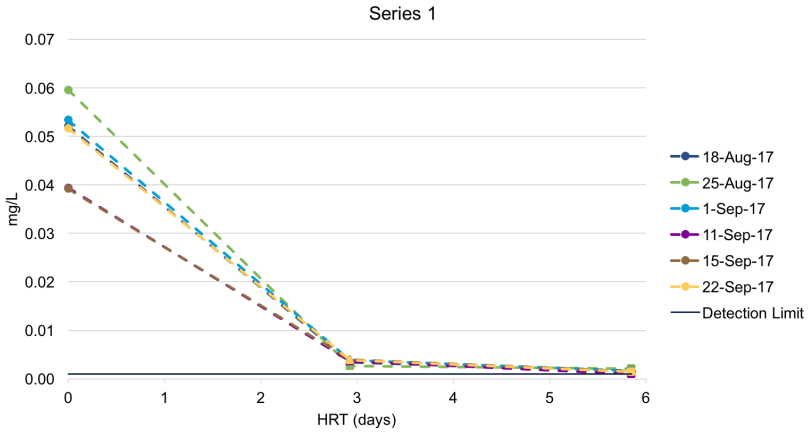
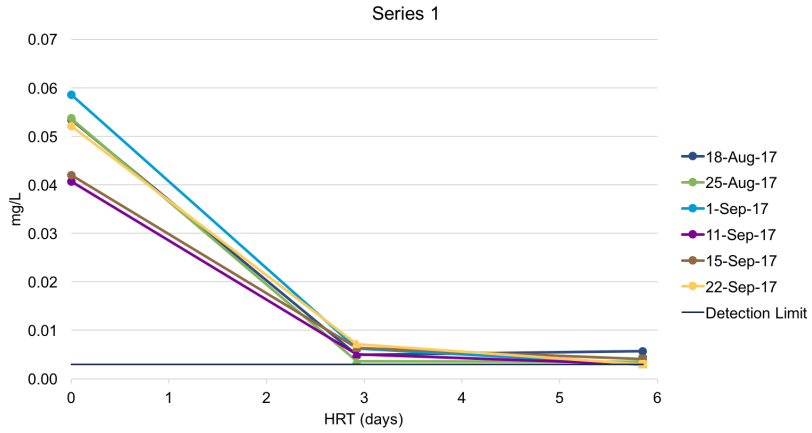
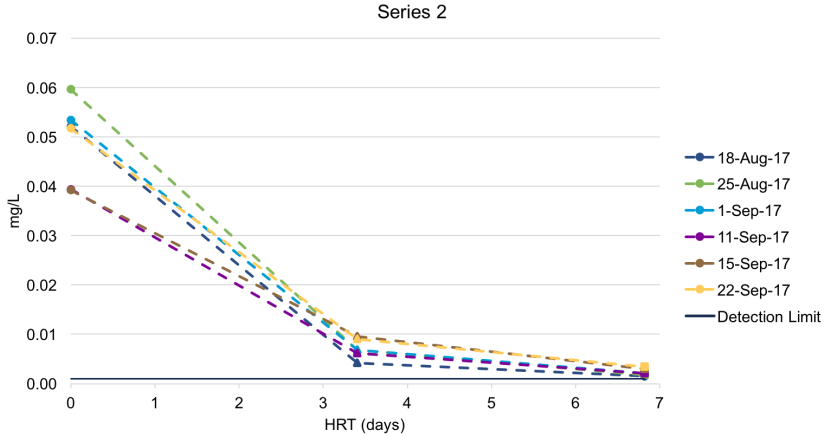


Figure B16 – Dissolved and total selenium concentrations through series 1 and series 2.
Dashed lines represent dissolved concentrations (left) while solid lines represent total concentrations (right). Circles represent the feed water entering the wetland, diamonds represent the outflow of cell A and squares represent the outflow of cell B. The ALS detection limit for selenium is 0.00005 mg/L.







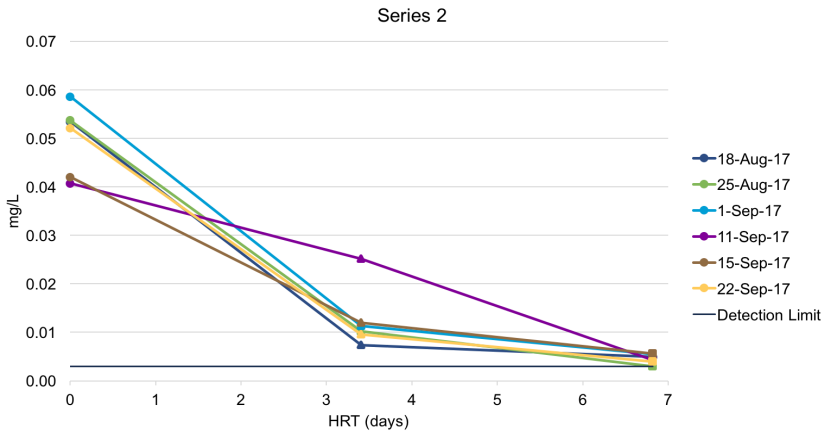


Figure B17 - Dissolved and total zinc concentrations through series 1 and series 2.
Dashed lines represent dissolved concentrations (left) while solid lines represent total concentrations (right). Circles represent the feed water entering the wetland, diamonds represent the outflow of cell A and squares represent the outflow of cell B. The ALS detection limit for dissolved zinc is 0.001 mg/L. The detection limit for total zinc is 0.003 mg/L.

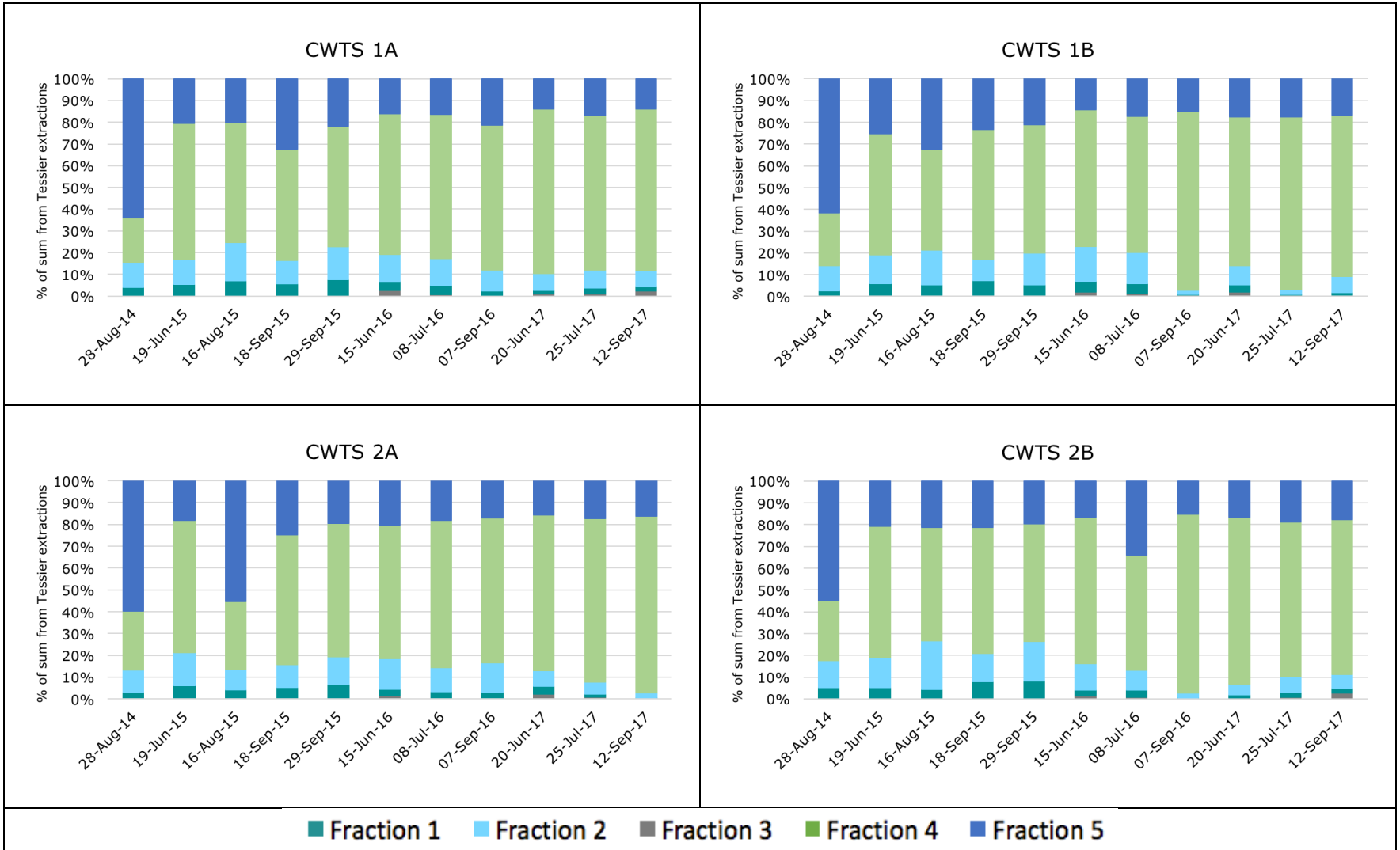


Figure B18 – Fraction distribution of copper expressed as percent over time (2014 – 2017).
Fractions are defined in Table B1.

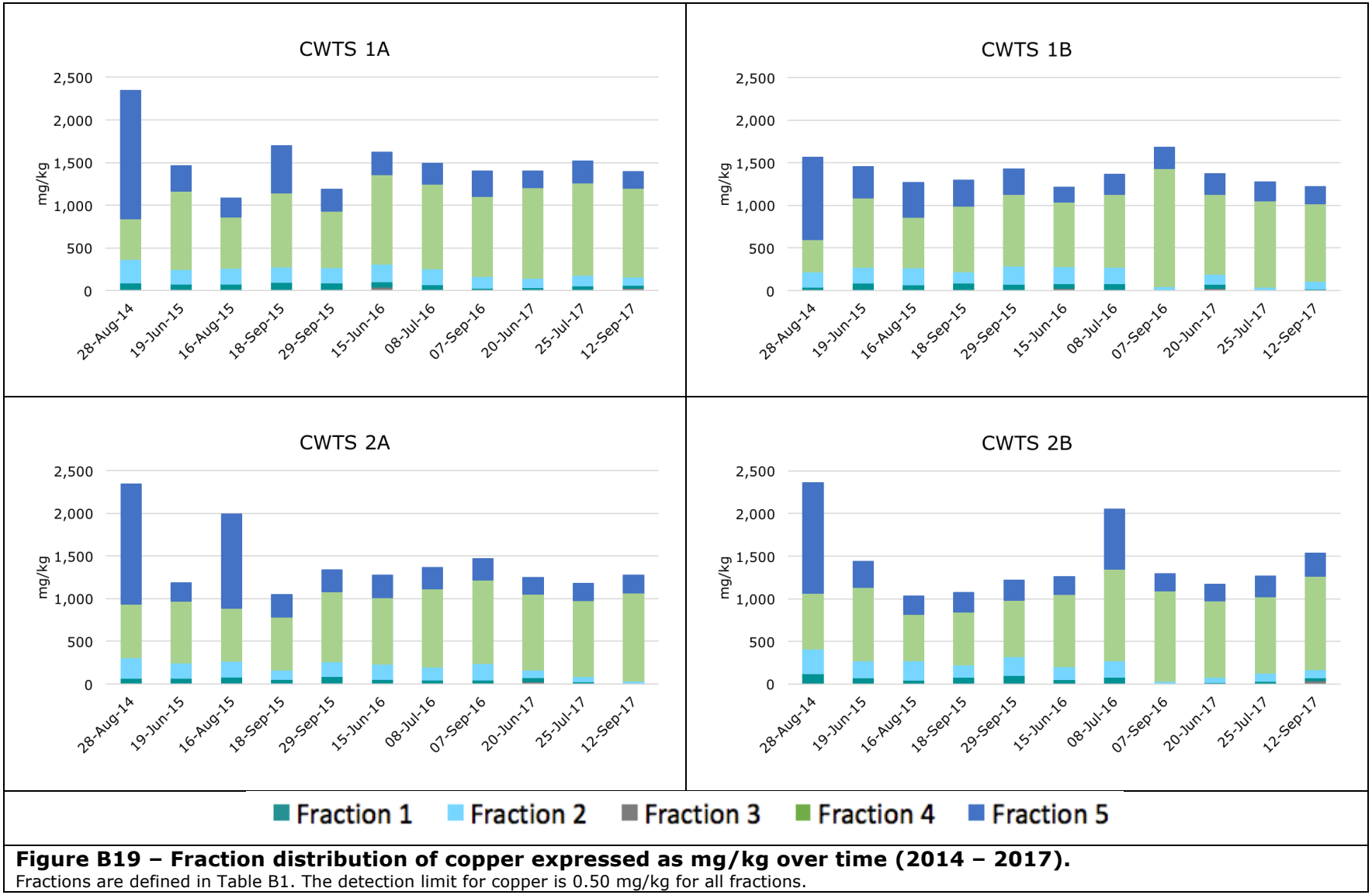


Figure B19 – Fraction distribution of copper expressed as mg/kg over time (2014 – 2017).
Fractions are defined in Table B1. The detection limit for copper is 0.50 mg/kg for all fractions.

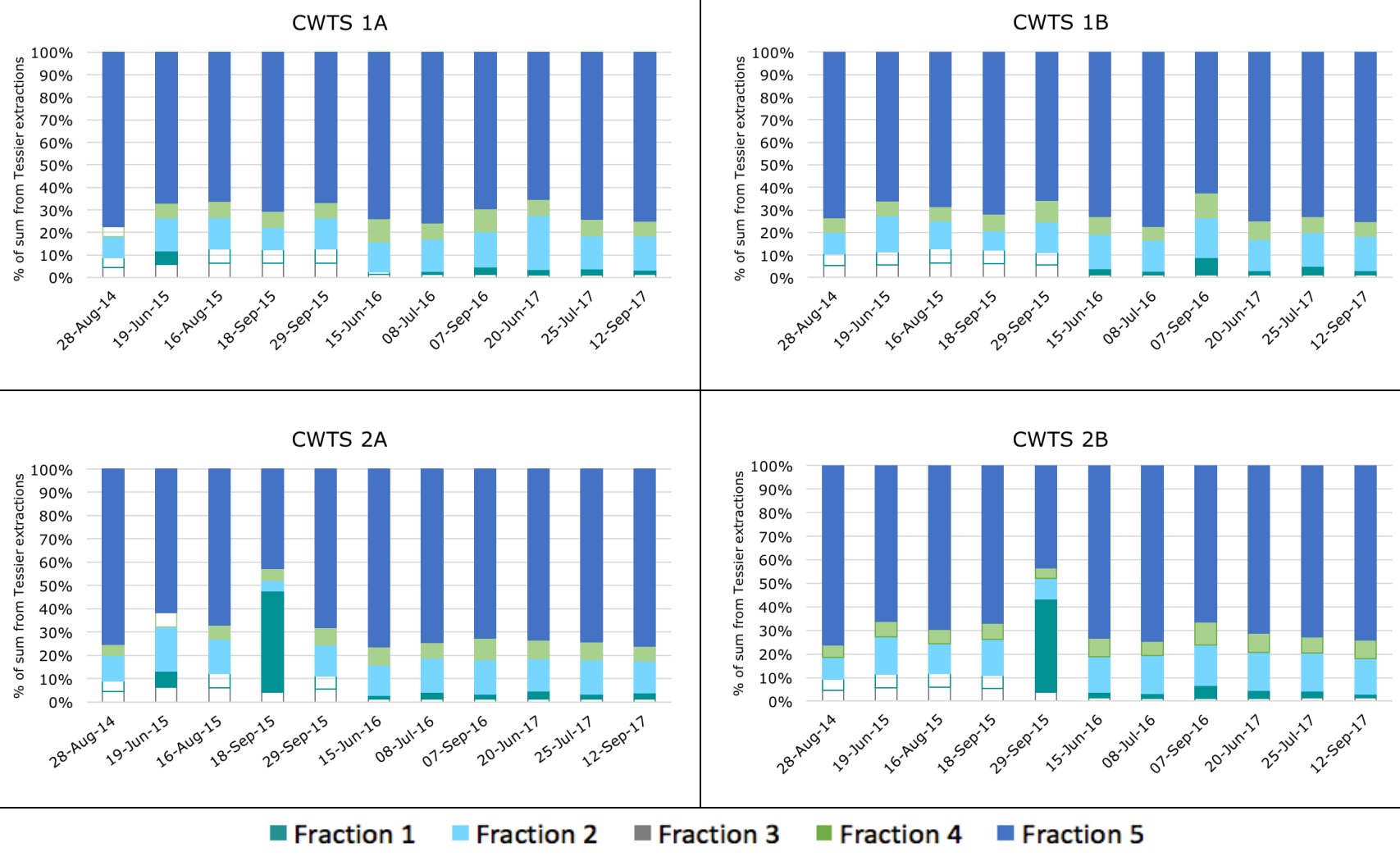


Figure B20 – Fraction distribution of zinc expressed as percent over time (2014 – 2017).
Unfilled areas are values at or below detection limits. Fractions are defined in Table B1.

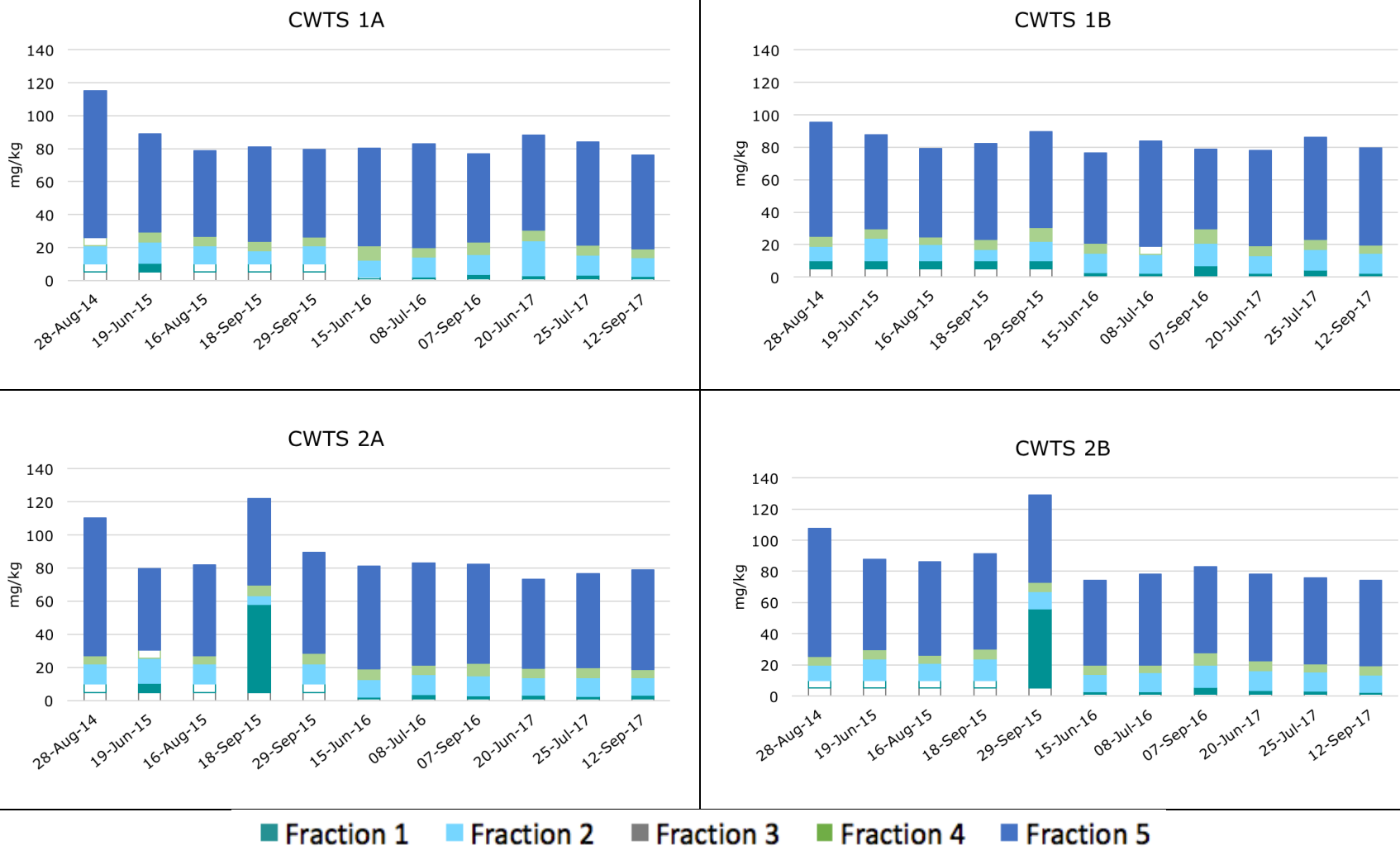


Figure B21 – Fraction distribution of zinc expressed as mg/kg over time (2014 – 2017).
A cells and B cells are on different y-axes. Unfilled areas are values at or below detection limits. Fractions are defined in Table B1. The detection limit for zinc is 1.0 mg/kg for all fractions.

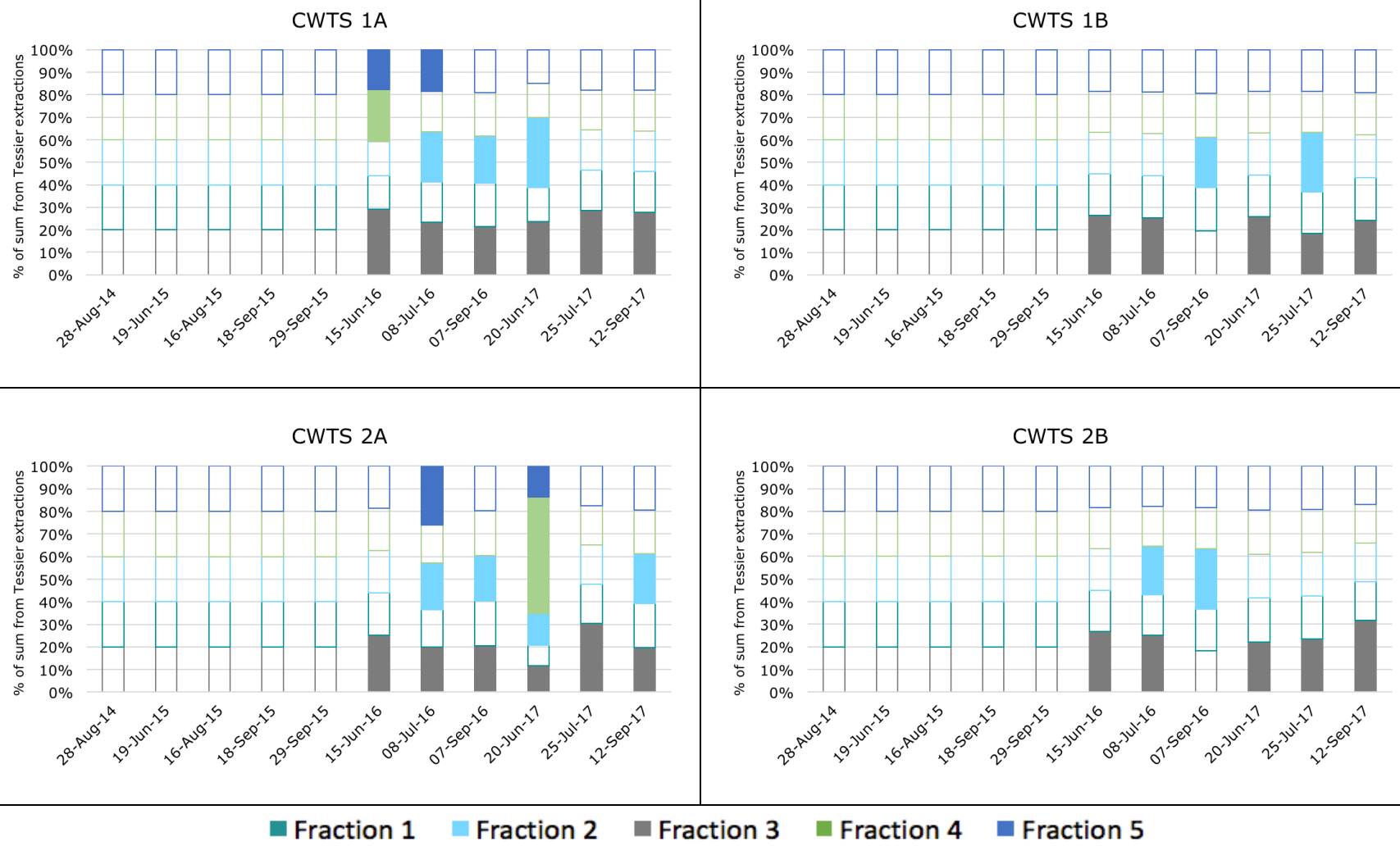


Figure B22 – Fraction distribution of cadmium expressed as percent over time (2014 – 2017).
Unfilled areas are values at or below detection limits. Fractions are defined in Table B1.

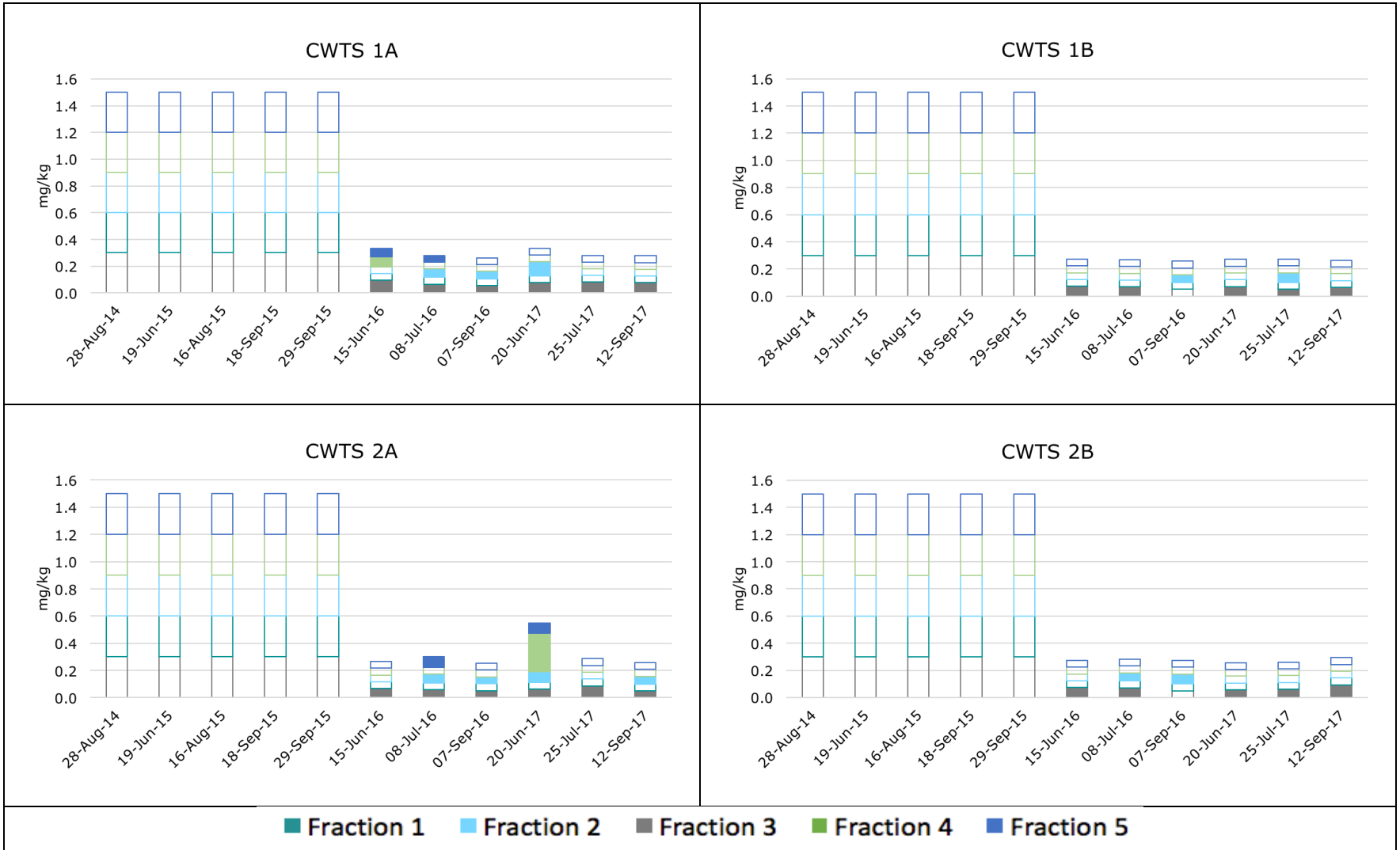


Figure B23 – Fraction distribution of cadmium expressed as mg/kg over time (2014 – 2017).
Unfilled areas are values at or below detection limits. Detection limits changed in 2016 and 2017, which is indicated by the smaller bars after 2015. The detection limit for cadmium in 2014 and 2015 was 0.30 mg/kg for all fractions. The detection limit for cadmium in 2016 and 2017 is 0.050 mg/kg for all fractions. Fractions are defined in Table B1.

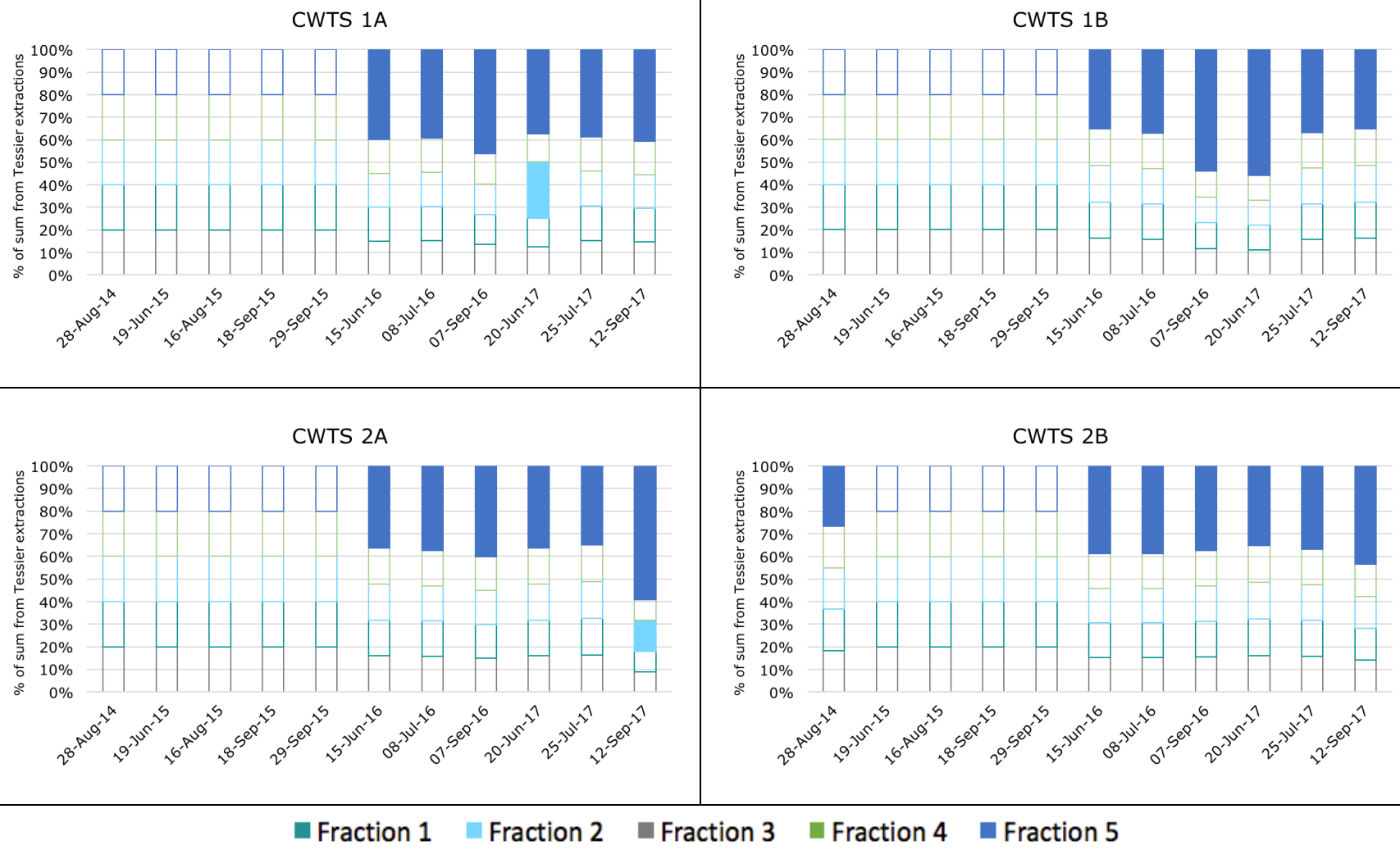


Figure B24 – Fraction distribution of molybdenum expressed as percent over time (2014 – 2017).
Unfilled areas are values at or below detection limits. Fractions are defined in Table B1.

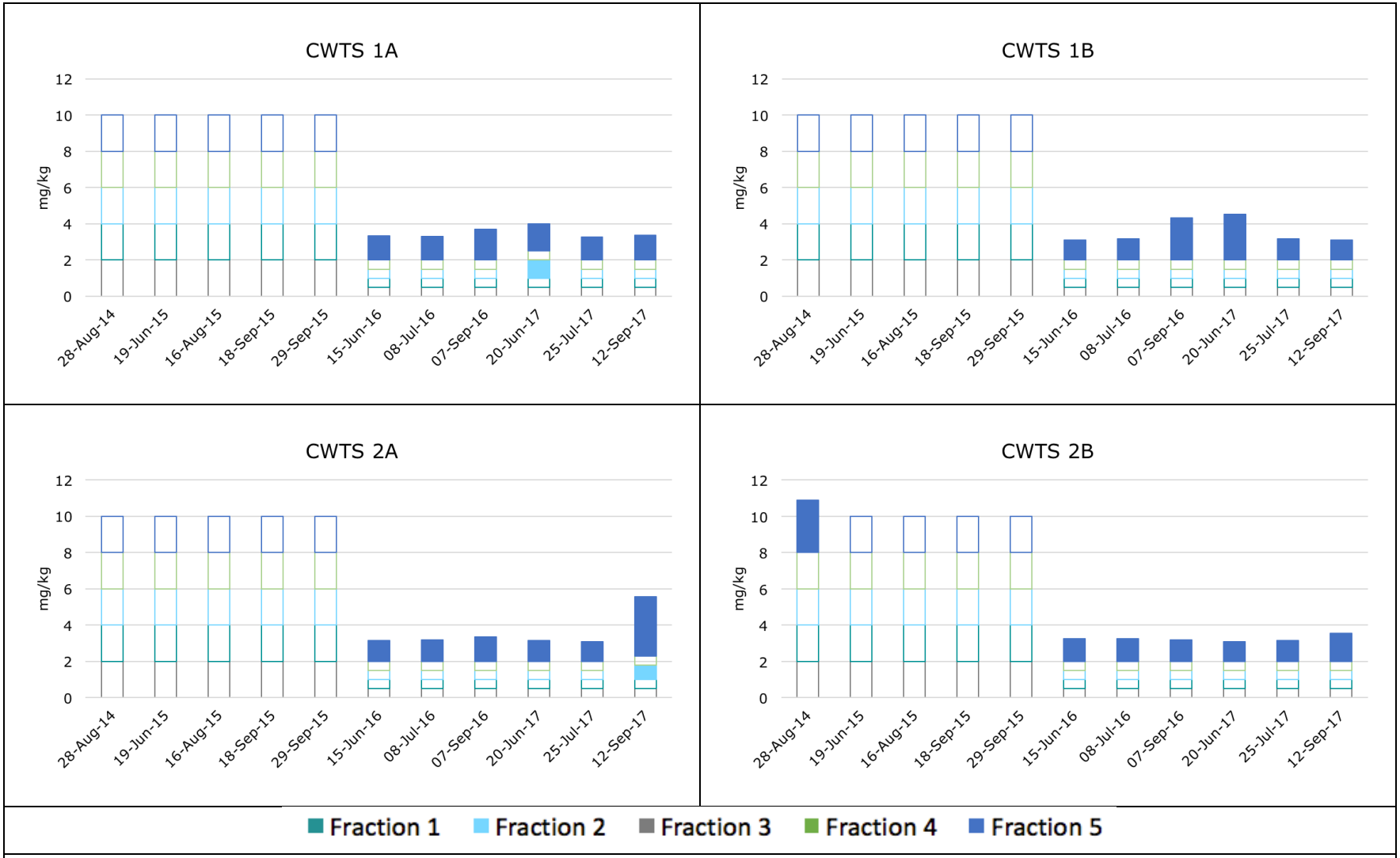


Figure B25 – Fraction distribution of molybdenum expressed as mg/kg over time (2014 – 2017).
Unfilled areas are values at or below detection limits. Fractions are defined in Table B1. Detection limits changed in 2016 and 2017, which is indicated by the smaller bars after 2015. The detection limit for molybdenum in 2014 and 2015 was 2.0 mg/kg for all fractions. The detection limit for molybdenum in 2016 and 2017 is 0.50 mg/kg for all fractions.

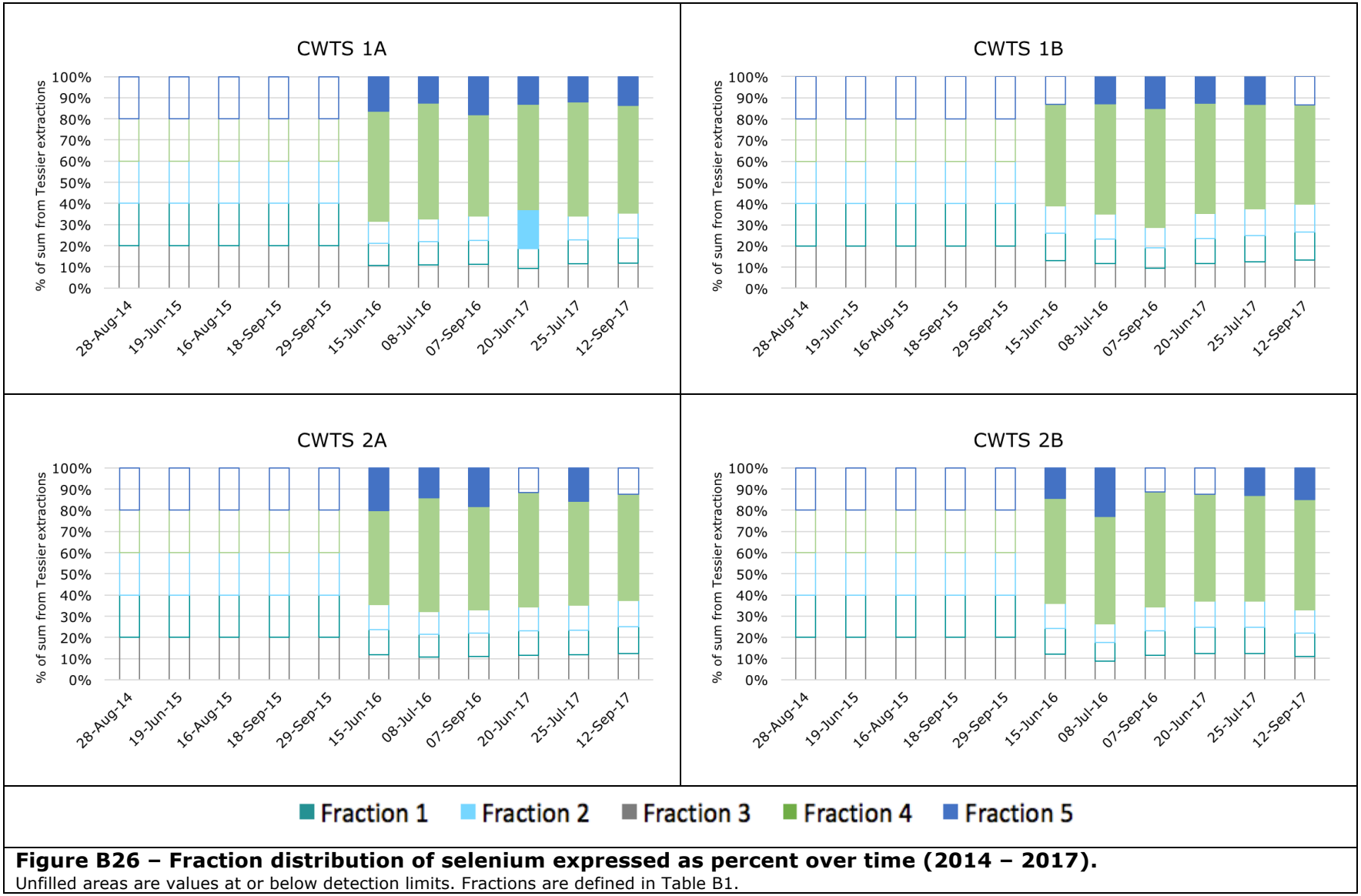


Figure B26 – Fraction distribution of selenium expressed as percent over time (2014 – 2017).
Unfilled areas are values at or below detection limits. Fractions are defined in Table B1.

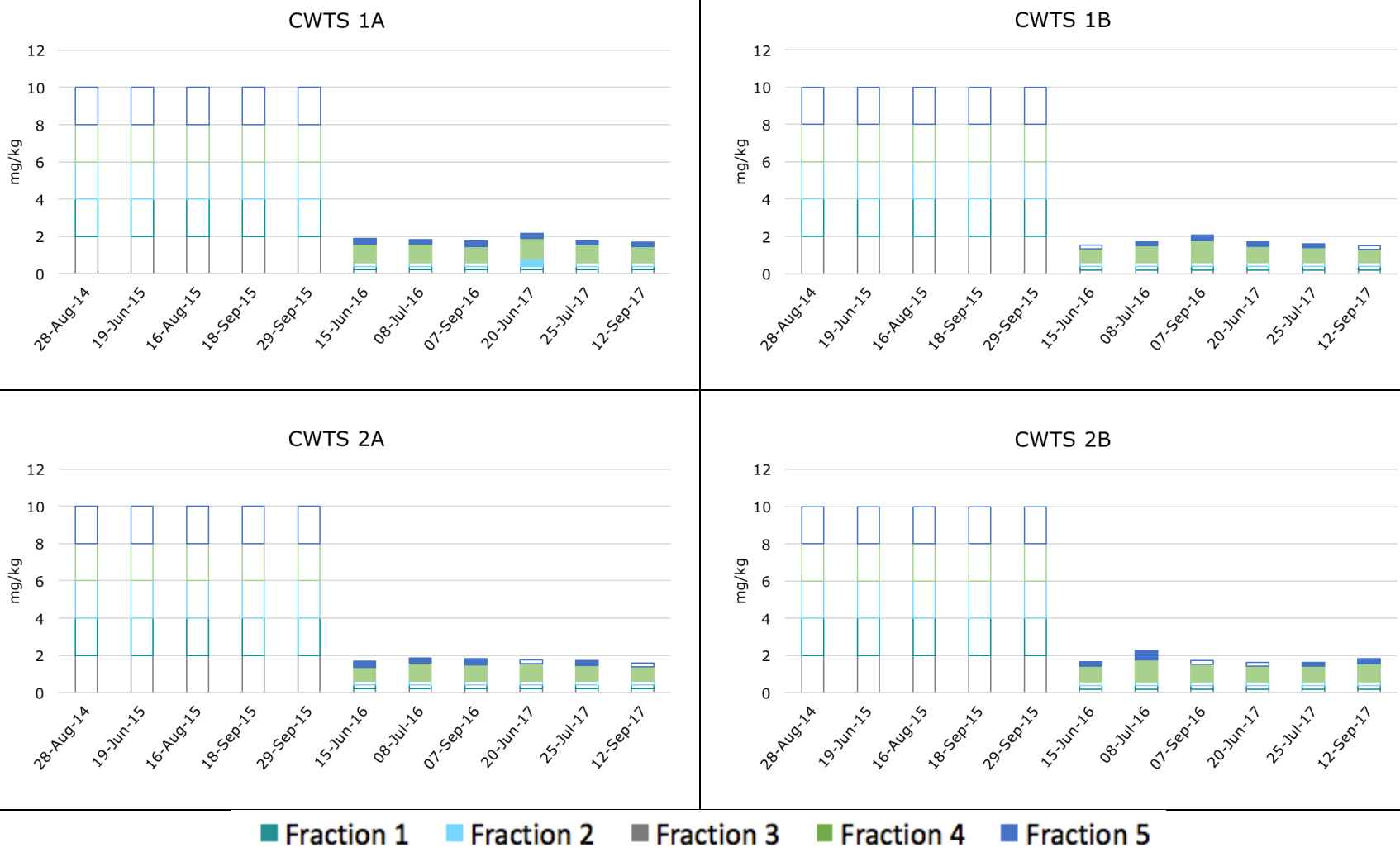


Figure B27 – Fraction distribution of selenium expressed as mg/kg over time (2014 – 2017).
Unfilled areas are values at or below detection limits. Detection limits changed in 2016 and 2017, which is indicated by the smaller bars after 2015. The detection limit for selenium in 2014 and 2015 was 2.0 mg/kg for all fractions. The detection limit for selenium in 2016 and 2017 is 0.20 mg/kg for all fractions. Fractions are defined in Table B1.

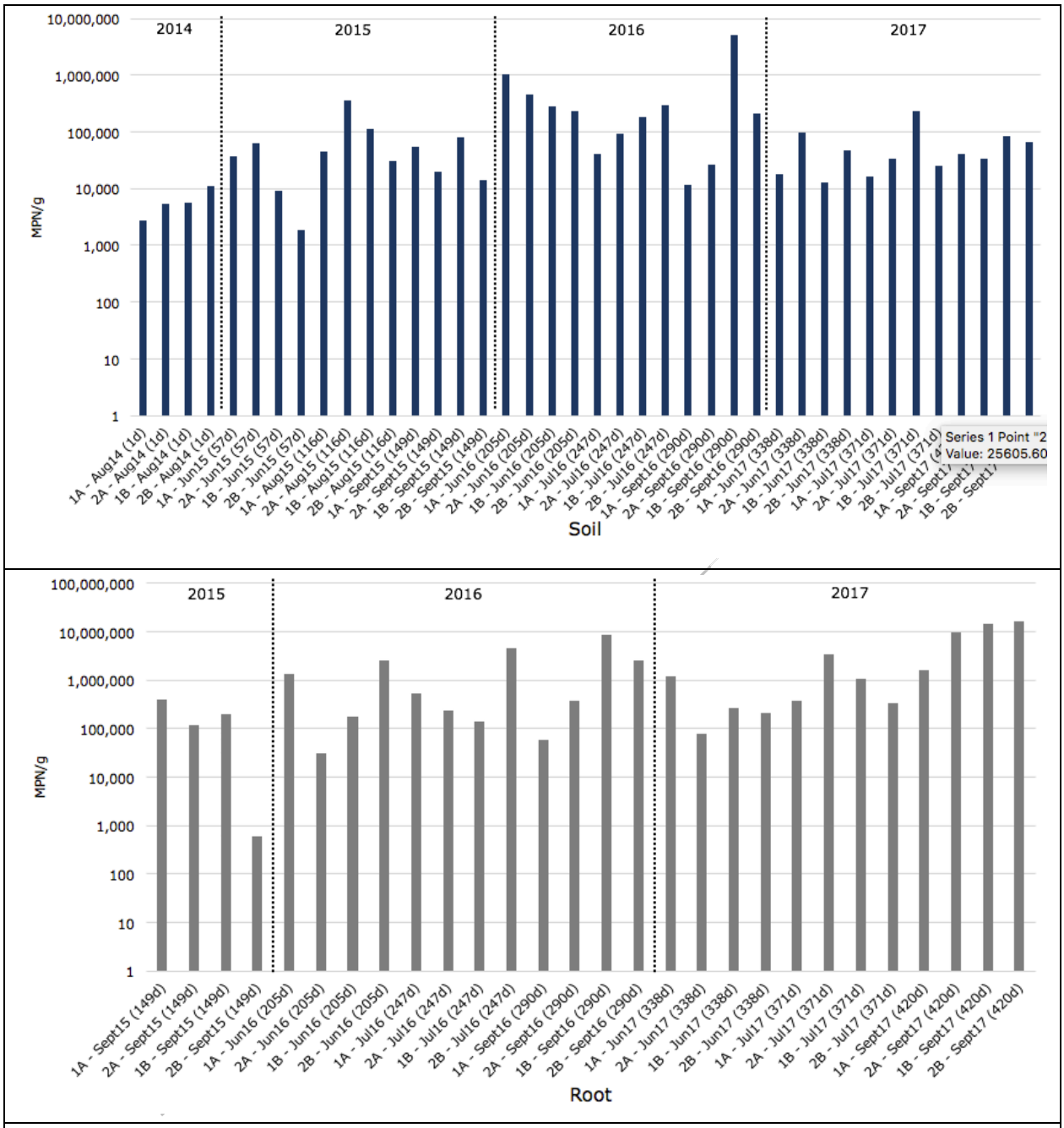
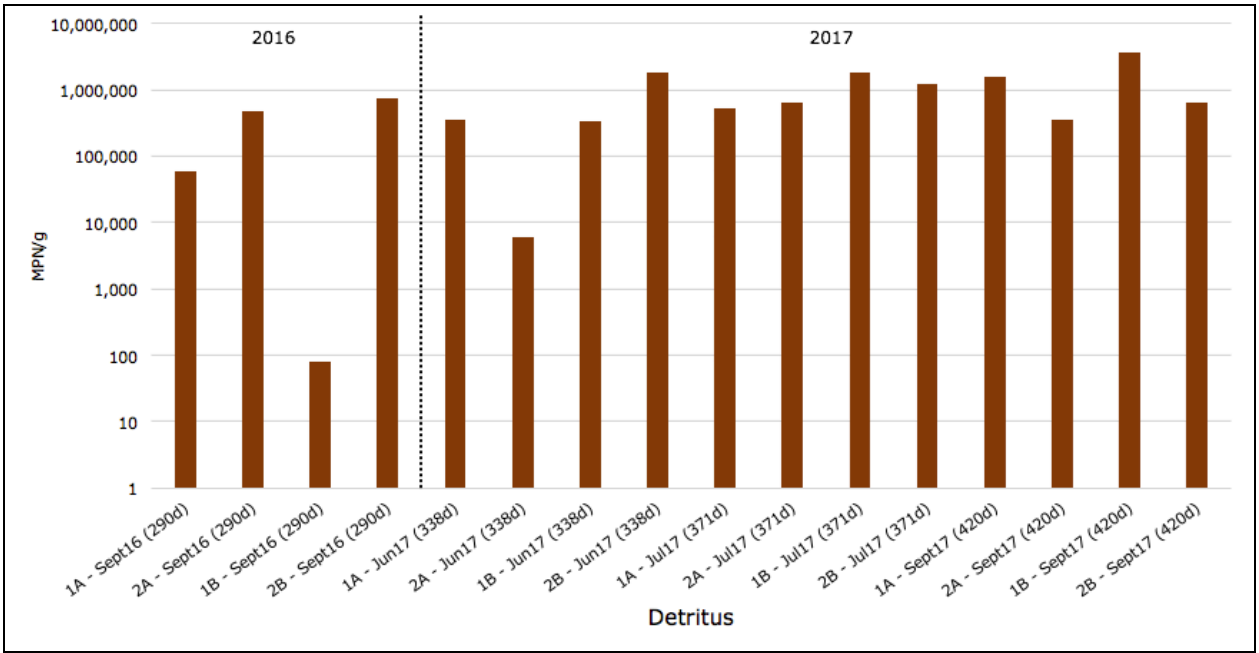


Figure B28 – Inferred abundance of sulphide-producing bacteria in soil (top) and roots (bottom) of the demonstration-scale CWTS.

Inferred abundance is calculated based on a combination of genetic and growth-based data. Shallow soil (0-5 cm); deep soil (10-20 cm); soil at 0d was that used for construction of CWTS. The root y-axis is different from the soil y-axis as the inferred abundance is much higher in root.



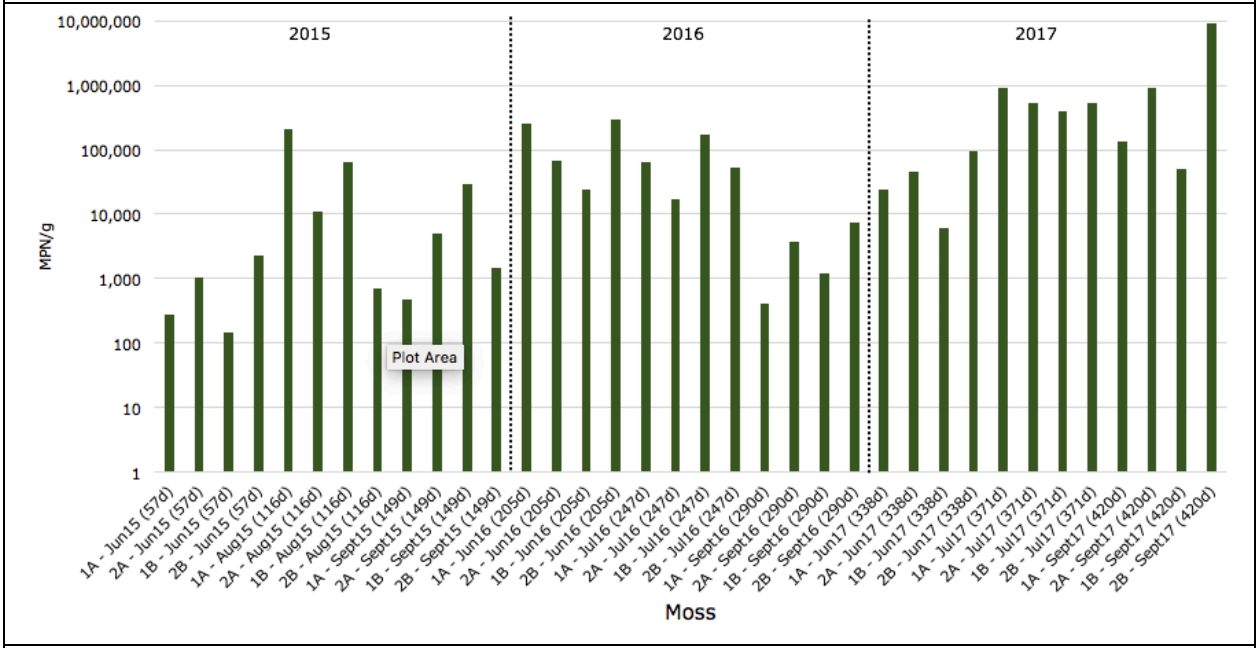


Figure B29 – Inferred abundance of sulphide-producing bacteria in detritus (top) and moss (bottom) of the demonstration-scale CWTS.
Inferred abundance is calculated based on a combination of genetic and growth-based data. Shallow soil (0-5 cm); deep soil (10-20 cm); soil at 0d was that used for construction of CWTS.

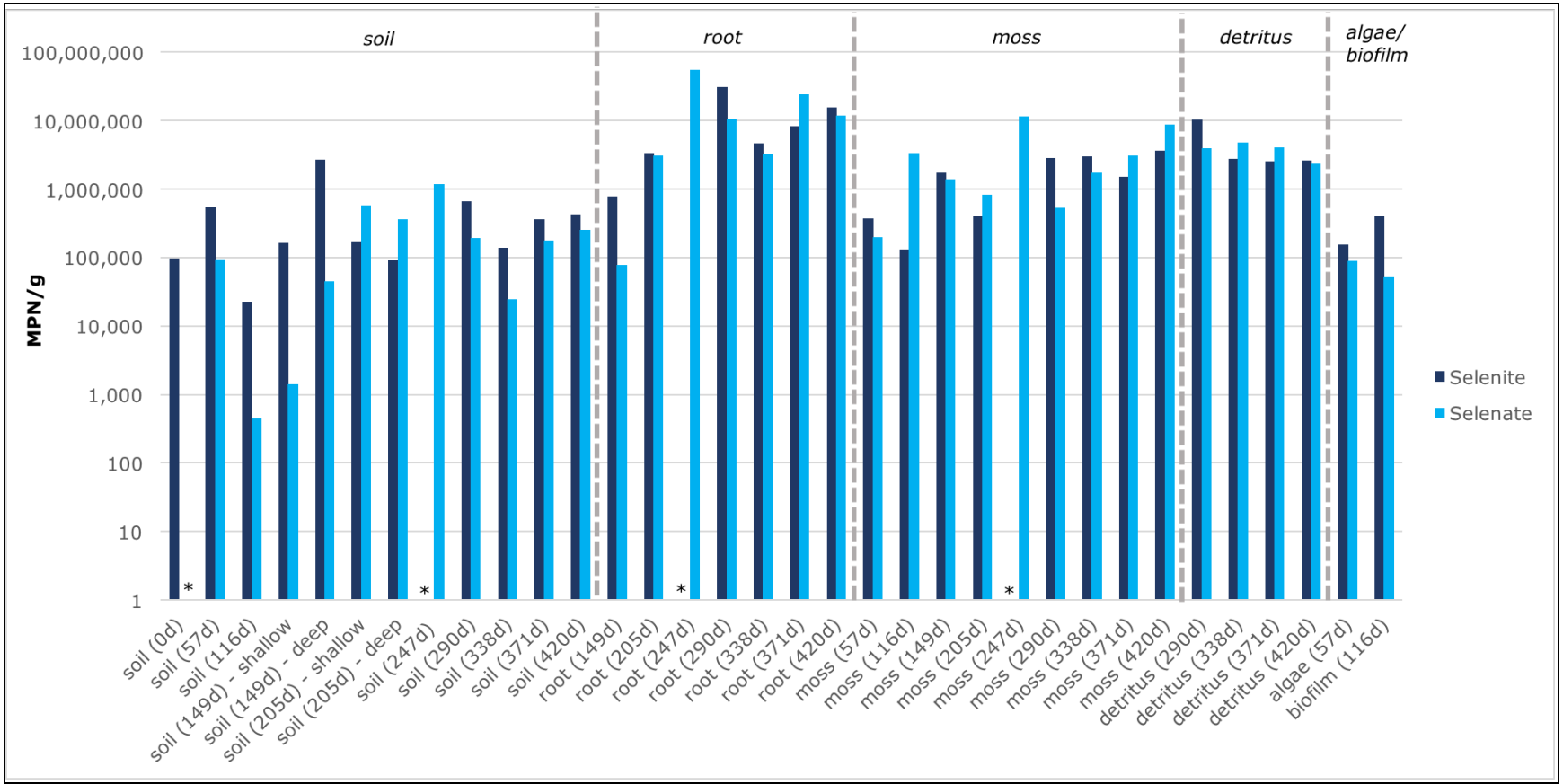


Figure B30 – Abundance of selenite- and selenate-reducing organisms in various CWTS sample types over time.
Shallow soil (0-5 cm); deep soil (10-20 cm); soil at 0d was that used for construction of CWTS. Organics material was added to the CWTS after day 236.

Appendix C

Table C1 - Schedule of work and activities for demonstration-scale CWTS

Item	Date	Activities	Actual
Construction	June 1-14 2014	Identify potential location for demonstration scale CWTS (Contango site visit – 1 scientist)	Completed
	June - July 2014	Engineering and geotechnical (Minto)	Completed
	July 2014	Construction (Minto)	Completed
	August 2014	Planting and bringing system online (Contango site visit – 1 scientist, 1 technologist), coordinate for local students to assist	Completed (no students available, brought 2 technologists)
Commissioning	2014	Acclimation and maturation at constant flow rate, ~20 hr HRT	Completed
		September - Contango site visit/checkup (1 technologist, 1 scientist)	Did not occur because construction was last week of August
	2015	Continued commissioning. Operation at constant flow rate, ~20 hr HRT	Completed (at shorter HRT)
		Spring – Contango site visit/checkup (1 technologist, 1 scientist), includes micro sampling	Completed
		Summer - Increase depth from 10 cm to 20 cm (1 technologist), includes micro sampling	Completed (scientist)
		Fall – Contango site visit/checkup (1 technologist), includes micro sampling	Completed (scientist)
	2016	Minto to add sandbags prior to first site visit and begin W15 creek monitoring	Completed May/June 2016
		Spring – Contango site visit (1 scientist, 1 technologist), includes microbial sampling and tasks outlined in Table 5 of report. HRT tracer study completed, outlined in section 7 of report.	Completed June 2016
		Summer - Contango site visit/checkup (1 scientist, 1 technologist), includes microbial sampling and tasks outlined in Table 5 of report. Evapotranspiration study completed as outlined in section 9 of report. Organics were added to the CWTS as outlined in section 5.1 of report.	Completed July 2016
		Fall - Contango site visit/checkup (1 technologist), includes microbial sampling and tasks outlined in Table 5 of report	Completed September 2016

	2017	Continued commissioning operations at constant flow rate.	Completed May 2017
		Two evapotranspiration studies were completed	May/June 2017
		Spring – Contango site visit (1 scientist, 1 technologist), includes microbial sampling and tasks outlined in Appendix A of report. Completed HRT tracer study. Sample for AVS to assess if the added organics are having the desired effect.	Completed June 2017
		Summer - Contango site visit/checkup (1 scientist), includes microbial sampling and tasks outlined in Appendix A of report.	Completed July 2017. Seasonal water sampling was not completed due to flow interruptions. Seasonal water sampling was completed by Minto in August.
Commissioning		Minto added sandbags to outflows of CWTS cells to increase water depth	August 2017
		Commissioning completed and beginning of operational period	Completed August 2017
Operations		Fall – Contango site visit/checkup (1 scientist), includes microbial sampling and tasks outlined in Table 5 of report.	Completed by Minto, after being taught microbial sampling by Contango. No site visit needed by Contango.
		Monitor soil redox. Determine when it consistently reaches targeted range.	Completed throughout 2017
Performance monitoring	2018	Fluctuate flow rates, based on expected amount of water for full-scale wetland (scaled to size). Evaluate performance of wetland at different key periods.	Throughout 2018,
Operations		Develop a flow rate schedule, sampling plan, and aphid control and monitoring plan for 2018	March, 2018
		Start water flow to CWTS; add sand bags on top of existing sandbags and to edges and end of CWTS to increase water depth to ~20 cm; aphid control using more powerful Insecticide on a routine schedule	April/May 2018

		Spring – Contango site visit; monitor for presence of aphids, collect one bag of *C. aquatilis* and one bag of polyester filter foam from each CWTS cell	May/June 2018
		Summer – Contango site visit; conduct a vegetation plot harvest and assessment of biomass produced	July/August 2018
		Conduct evapotranspiration study for warmer months	July/August 2018
		Fall – Contango site visit (if necessary), replanting if needed.	September 12, 2018
		2018 Update Report	December 2018
Reporting	2014-2018	Reporting will be performed annually, with verbal and/or emailed interim updates	On Schedule

[1] The commissioning-B period will continue in 2017 until the demonstration-scale CWTS performance is consistently achieving performance expectations.

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Technical Memo

To: Ryan Herbert, Environmental and Permitting Manager, Capstone Mining Corp. Minto Mine

From: Monique Simair & Rachel Martz, Contango Strategies Ltd.

Date: December 2017

Project: Minto Explorations Ltd. - Future Terraced Wetland (FTW) area near W15

Client: Capstone Mining Corp.

Subject: 2017 Future Terraced Wetland Area Monitoring Program

Document #: 011_1217_09B

1. Introduction

Minto Mine's Reclamation and Closure Plan (2017-01) includes consideration for a future terraced wetland (FTW) with in the upper south-west reach of the Minto Creek watershed to treat water that is impacted by the Southwest Waste Rock Dump (SWD) seepage. Therefore, Contango was contracted in 2016 to complete an assessment of the FTW area to determine if a terraced wetland was practicable for improving water quality at the W15 pond (collection point for the watershed). Contango completed an initial study of the FTW area in 2013 during a site assessment (Contango, 2014). In 2016 Contango implemented a monitoring program to define the seepage inflow path to the FTW area, and assess water quality over time (Contango, 2017a). This current memo is an update to the FTW monitoring program, with 2017 data incorporated. These findings are guiding an options assessment of the FTW area.

1.1 2013 Study Findings

In 2013, a site assessment was completed by Contango Strategies Ltd. (Contango, 2014) to inform the design of site-specific passive treatment technologies for the Minto site post closure. One of the primary efforts of the 2013 site assessment was to identify natural water treatment processes that are currently occurring at the Minto site. As such, the focus of the 2013 assessment was on the SWD drainage, which consists of SWD seepage, the upper reaches of Minto Creek and a seepage collection pond area known as "W15".

One of the primary findings of the 2013 assessment was that there were elevated concentrations of constituents at the W15 sampling location, and in the SWD seeps but concentrations were lower in the seepage fed creek upstream of W15. This suggested that natural water treatment was occurring between seeps along the toe of the SWD and the creek. The initial field investigation in 2013 suggested that the source of contaminants at W15 was from road runoff from a culvert entering the W15 area (Contango, 2014), which was later found to also include water from the W38 SWD seep. However, there was only one set of data from 2013 on which to make this inference so further studies were warranted.

1.2 2016 Study Findings

In June 2016, a monitoring program was developed and initiated on the components of the SWD drainage area. The monitoring program was intended to further investigate the findings of the 2013 site assessment and determine the source of impacted water to the W15 seepage

collection pond. The monitoring program selected five additional sampling sites (FTW-A, -B, -B1, -C, and –D) that intercepted locations where water was entering the W15 pond. FTW-A, -B, and -B1 were branches of the main creek that feed the W15 pond from the southwest, and FTW-C and -D were receiving seepage water from W38 (FTW-D had higher flows than FTW-C, and corresponded to the area identified in 2013 that also had culvert runoff). Based on 2016 findings it was concluded that FTW-B was representative of the water entering W15 from the creek to the southwest, and was low in copper concentrations. Meanwhile, high copper concentrations were found at FTW-D entering W15 from the west, and suggests that seepage from W38 is the major load contribution, rather than the creek coming from W10 to W15, which was originally the target for treatment by the FTW. Based on the results of the 2016 monitoring program (Contango, 2017), water quality monitoring was recommended to continue in 2017 to further assess these findings.

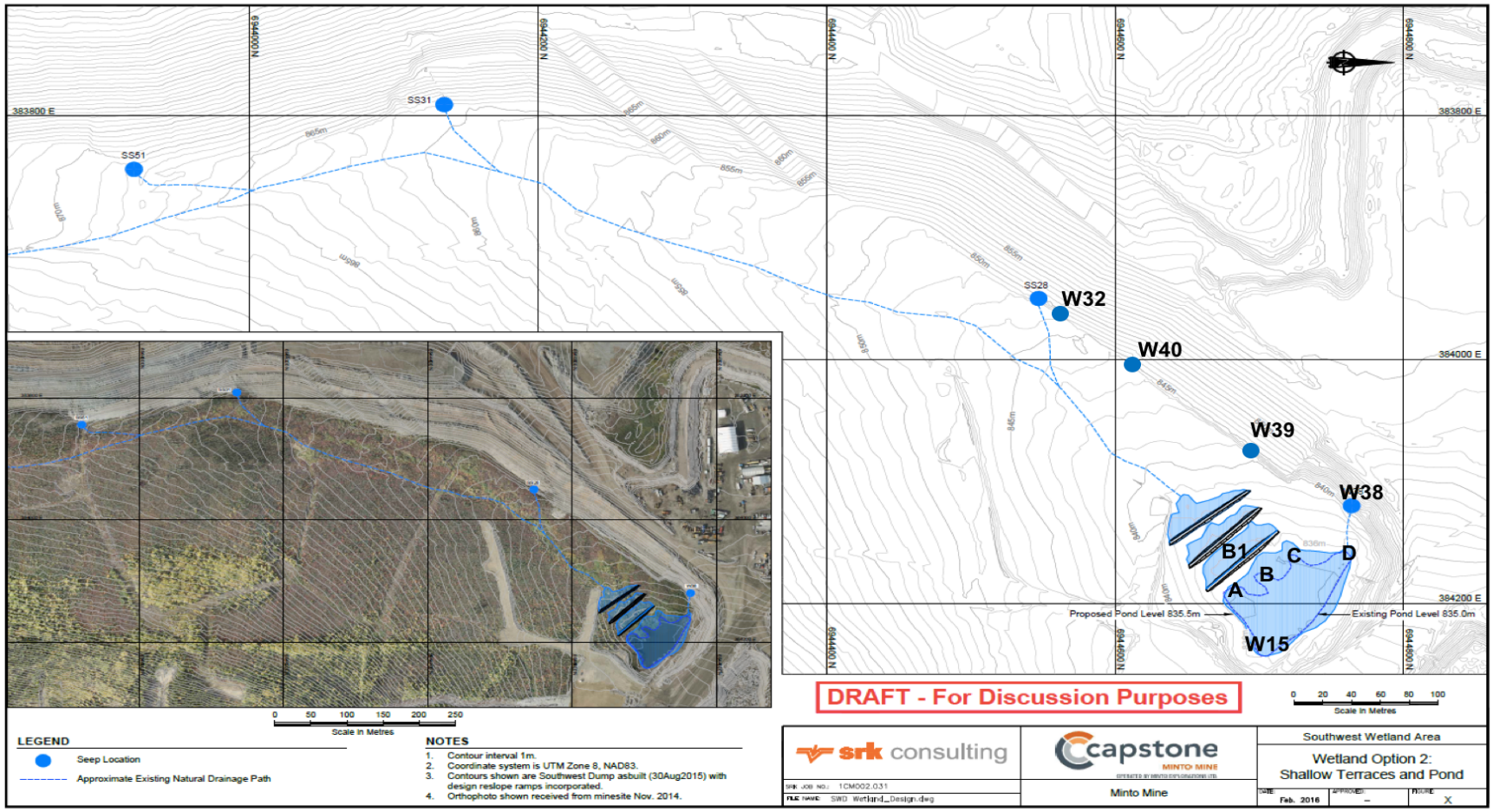


Figure 1 - Proposed terraced wetlands and sampling locations.
Estimated location of W32, W39, and W40 seeps shown on map. W39 is approximately 80 m southwest of W38 on the same toe of the SWD. Sites A, B, B1, C and D are equivalent to sites FTW-A, FTW-B, FTW-B1, FTW-C, FTW-D discussed in this report. W10 sampling location is approximately 1.5 km Southwest of W15.

1.3 2017 Study Objectives

The objective of the 2017 program was to continue monitoring and gathering additional data to more accurately determine the source and quality of impacted water to the W15 pond area, such that a terraced wetland or other treatment options could be properly selected, designed, and situated. Sampling sites were refined in 2017 to include FTW-B, FTW-D, W15, and surrounding seeps (i.e., removing FTW-A, -B1, and -C).

2. Methods

The 2017 FTW monitoring program consisted of water sampling at FTW-B, FTW-D (downstream of Seep W38), and W15 on a biweekly basis. Seeps W32, W38, W39, and W40 in the W15 area were also monitored on a biweekly basis in 2017, as part of the Minto Mine's routine monitoring program. Data from these seeps was also incorporated into the FTW monitoring program. The 2017 sampling program collected limited data from the seep locations as most of the seeps were dry (Table 1). FTW-D was also dry for most of 2017 which is expected as the flow at FTW-D generally appears to originate from seep W38 which was also dry.

Land reclamation on the SWD has been ongoing through this monitoring program and is intended to decrease seepage in the area, and therefore may be contributing to changing seepage patterns and diminishing the quantity of water that reaches the seeps. Additionally, 2017 had 24% less precipitation than 2016 with total rainfall between April and September in 2016 amounting to 223 mm and in 2017 amounting to 169 mm (weather data obtained from Minto).

Water samples for the FTW monitoring program have been analyzed for parameters in Table 2. Samples were collected by grab method, field filtered, and/or preserved as appropriate to the analysis. Following each sampling event, in situ measurements were taken using a handheld YSI unit at each sample location. *In situ* parameters measured include temperature, dissolved oxygen (DO), specific conductance (SPC), pH, and oxidation-reduction potential (ORP).

Dissolved metals concentrations were used in this report rather than total metals concentrations, as total values were variable. The variability of the total metals concentrations is likely owing to particulate material containing metals being part of the grab sample collection method (influenced by TSS) and are likely not representative of the dissolved bioavailable metals concentrations at the sampling locations.

Table 1 – Sample location monitoring.

Date	Sample Locations						
	W15	W32[1]	W38[1]	W39[1]	W40[1]	FTW-B	FTW-D[2]
26-May-17	Sampled	Sampled	Not Sampled[3]	Not Sampled[3]	Not Sampled[3]	Sampled	Sampled
09-Jun-17	Sampled	Sampled	Frozen	Dry	Dry	Sampled	Sampled
21-Jun-17	Sampled	Dry	Dry	Dry	Dry	Sampled	Dry
07-Jul-17	Sampled	Dry	Dry	Dry	Dry	Sampled	Dry
22-Jul-17	Sampled	Dry	Dry	Dry	Dry	Sampled	Dry
04-Aug-17	Sampled	Dry[4]	Dry	Dry	Dry[4]	Sampled	Dry
11-Aug-17	Not Sampled[3]	Dry	Dry[5]	Dry[5]	Dry	Not Sampled[3]	Dry[5]
01-Sep-17	Sampled	Dry	Dry	Dry	Dry	Sampled	Dry
11-Sep-17	Sampled	Dry	Dry	Dry	Dry	Sampled	Dry
29-Sep-17	Sampled	Dry	Dry	Dry	Dry	Sampled	Dry

[1] Seeps were often dry in 2017 due to land reclamation in the area and low precipitation.
[2] FTW-D is fed by W38 (which was often dry), resulting in limited samples from FTW-D.
[3] Sampling events missed.
[4] Site not visited due to high risk bear activity in the area. It was assumed these sites were dry due to the proximity of the seep to other seep locations that were dry.
[5] Gaps in record keeping were due to human error. It was assumed these sites were dry due to the proximity of the seep to other seep locations that were dry.

Table 2 – 2017 water sampling and analysis program.

Sample location	Frequency	Water Sample Parameters
W15	Biweekly	Total metals Dissolved metals Routine (Ca, Mg, Na, K, Cl, SO_4, NO_3, NO_2, hardness, Alkalinity, pH, EC, TDS) TSS COD TKN Ammonia TOC Total Nitrogen
W32		
W38		
W39		
W40		
FTW-B		
FTW-D		

3. Results

Seepage from W32 appears to flow towards the creek before sampling site FTW-B, but was not observed to reach the creek by surface flow. That is, the seepage water daylights at the toe of the SWD, but infiltrates to the ground and did not appear to surface before the creek. There appears to be natural attenuation (e.g., treatment or dilution) occurring between these seeps and the creek (as measured at FTW-B). This is suggested with the lower concentrations of all constituents of concern (cadmium, copper, molybdenum, selenium, nitrate, nitrite) at FTW-B compared the seeps (Table 3, Figure 2 to Figure 7). Additionally, SPC was higher at W32, but lower at FTW-B, reaffirming that natural treatment or dilution is occurring between W32 and FTW-B (Table 4). Moreover, FTW-B had lower concentrations of copper, molybdenum, and nitrate (with the exception of the first sampling event in 2017 for nitrate) and similar concentrations of cadmium and selenium compared to W15 throughout 2016 and 2017 (Table 3, Figure 2 to Figure 7).

The lower concentrations observed at FTW-B compared to W15 confirms that FTW-B and contributing seeps are not the source of elevated constituents in the W15 pond (Table 3, Figure 2 to Figure 7). The constituents of concern must therefore enter the W15 pond from another source. Following the 2016 monitoring program, seeps feeding FTW-D (e.g., seep W38) were hypothesized to be the source of elevated concentrations of constituents at W15 (Contango, 2017a). The 2016 findings showed high copper concentrations at FTW-D (fed by W38) compared to W15 (Table 3, Figure 2). The 2017 sampling also indicated elevated concentrations of all constituents of concern (except nitrate) at FTW-D compared to W15, further suggesting that W38 (which feeds FTW-D) is the source of elevated metals and metalloids at W15 (Table 3, Figure 2 to Figure 7).

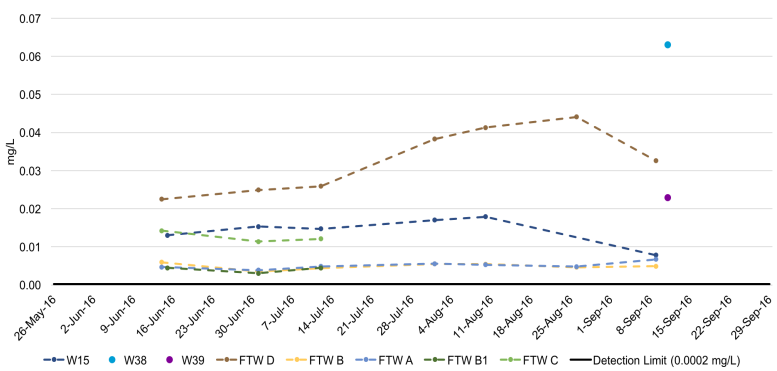
Annual trends at W15 water chemistry for copper have remained consistent since 2014. This is notable because in 2016 W38 flowed year-round while in 2017 W38 was mostly dry. Although there was no surface water flow at FTW-D in 2017, it may contribute constituents to W15 through subsurface flow. This is supported by the similar copper concentrations observed throughout the monitoring program, as well as the seasonal trends occurring each year (Figure 8). Alternatively, it is possible that there might be elevated constituents in the soils and sediments of W15 pond, releasing metals and metalloids into the water as has been seen with the copper and aluminum bearing soils that were used in construction of the CWTS on site (Contango, 2015; 2016; 2017b).

Reclamation activities and lower precipitation appear to be the major contributing factors in decreased visible seepage water flows along the toe of the SWD. Reclamation activities over the past two years include re-contouring and overburden placement in the SWD. Annual precipitation in 2017 decreased compared to 2016. Planned reclamation activities for the SWD, such as covers and vegetation, are expected to improve water quality and decrease volumes of water reporting from the seepage locations.

Table 3 – FTW 2017 sampling program average, minimum, and maximum concentrations for constituents of concern.

Parameter (mg/L)		W15		W32		FTW-B		FTW-D	
		2016	2017	2016	2017	2016	2017	2016	2017
Cd	Min	0.000005	0.000005	N/A	0.000019	0.000005	0.000005	0.000008	0.000021
	Avg	0.000007	0.000006	N/A	0.000039	0.000010	0.000006	0.000020	0.000033
	Max	0.000011	0.000009	N/A	0.000059	0.000024	0.000012	0.000038	0.000046
Cu	Min	0.0078	0.0088	N/A	0.0208	0.0036	0.0037	0.0225	0.0368
	Avg	0.0143	0.0192	N/A	0.0293	0.0049	0.0082	0.0328	0.0449
	Max	0.0179	0.0286	N/A	0.0377	0.0060	0.0230	0.0441	0.0529
Mo	Min	0.00248	0.00194	N/A	0.00647	0.00216	0.00193	0.00243	0.00490
	Avg	0.00340	0.00257	N/A	0.01214	0.00264	0.00235	0.00509	0.00696
	Max	0.00395	0.00319	N/A	0.01780	0.00319	0.00289	0.00723	0.00902
Se	Min	0.0023	0.0013	N/A	0.0125	0.0026	0.0014	0.0020	0.0007
	Avg	0.0038	0.0018	N/A	0.0159	0.0033	0.0019	0.0064	0.0018
	Max	0.0047	0.0028	N/A	0.0193	0.0041	0.0023	0.0082	0.0029
NO_3	Min	8.8	4.9	N/A	33.8	9.0	7.2	14.0	0.5
	Avg	14.8	7.3	N/A	40.9	12.7	9.3	24.2	4.4
	Max	19.1	9.8	N/A	47.9	17.4	11.5	29.0	8.2
NO_2	Min	0.0093	0.0062	N/A	0.2220	0.0056	0.0036	0.0020	0.0950
	Avg	0.0261	0.0382	N/A	0.3785	0.0155	0.0607	0.0095	0.2135
	Max	0.0375	0.1580	N/A	0.5350	0.0239	0.4430	0.0264	0.3320

N/A = location was not sampled in 2016 as the seep was dry. W32 and FTW-D have recorded values from 26-May-17 and 09-Jun-17. W15 and FTW-B have recorded values from 26-May-17 to 29-Sep-17.



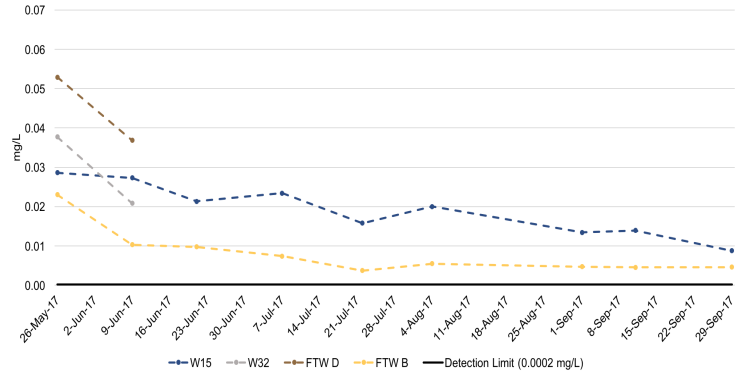
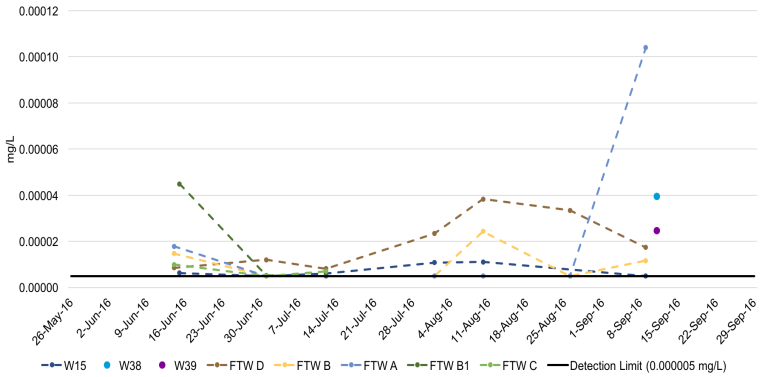


Figure 2 - Concentrations of dissolved copper at the W15 pond area.
The left graph shows 2016 data and the right graph shows 2017 data. W38 and W39 were sampled once in 2016 and are represented by the single points.



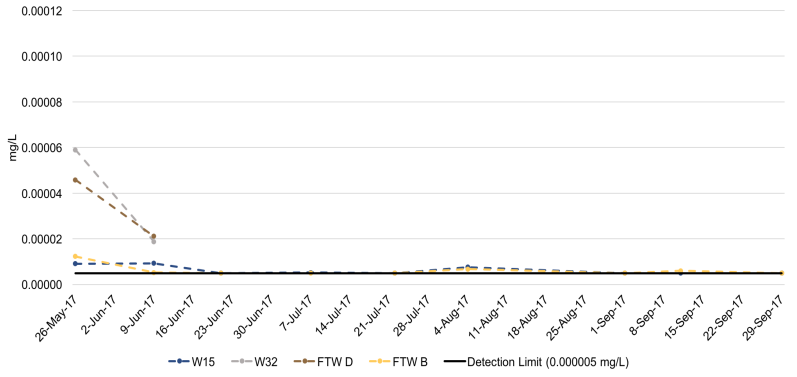
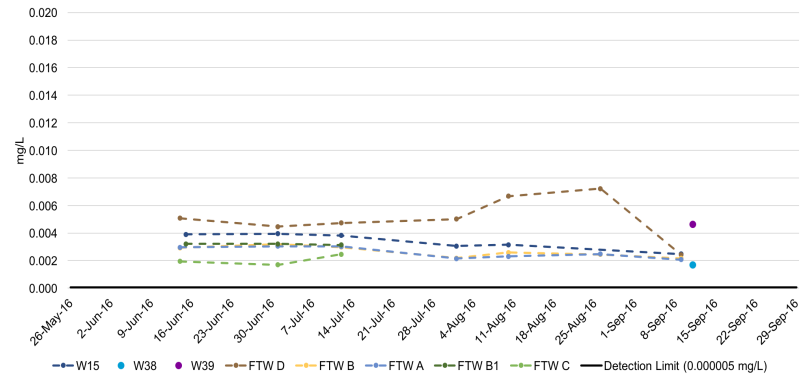


Figure 3 – Concentrations of dissolved cadmium at the W15 pond area.
The left graph shows 2016 data and the right graph shows 2017 data. W38 and W39 were sampled once in 2016 and are represented by single points.



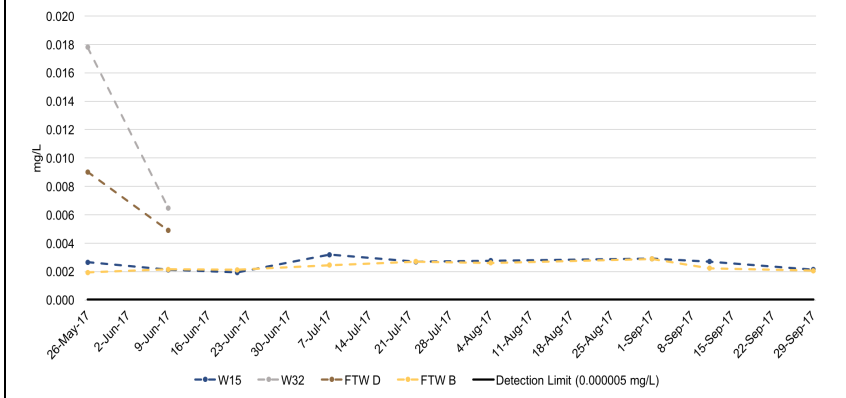
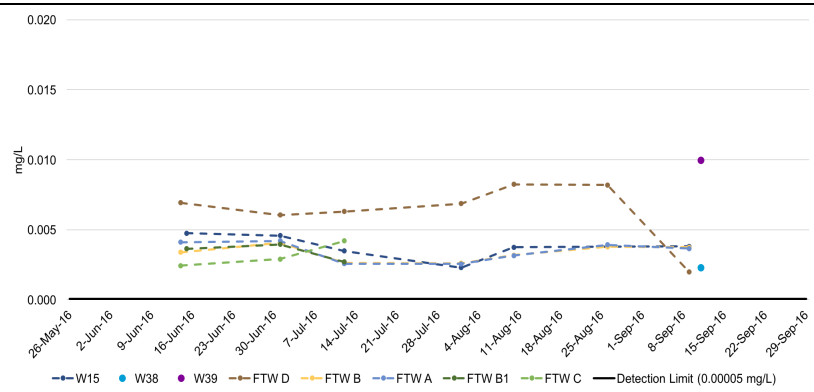


Figure 4 – Concentrations of dissolved molybdenum at the W15 pond area.
The left graph shows 2016 data and the right graph shows 2017 data. W38 and W39 were sampled once in 2016 and are represented by single points.



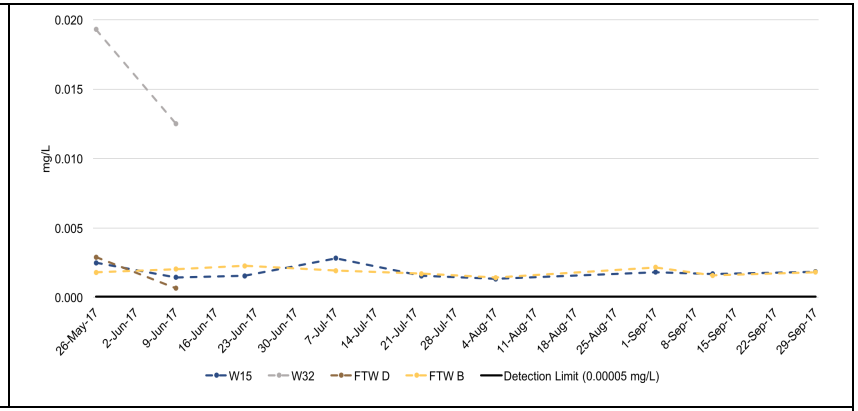


Figure 5 – Concentrations of dissolved selenium at the W15 pond area.
The left graph shows 2016 data and the right graph shows 2017 data. W38 and W39 were sampled once in 2016 and are represented by single points.

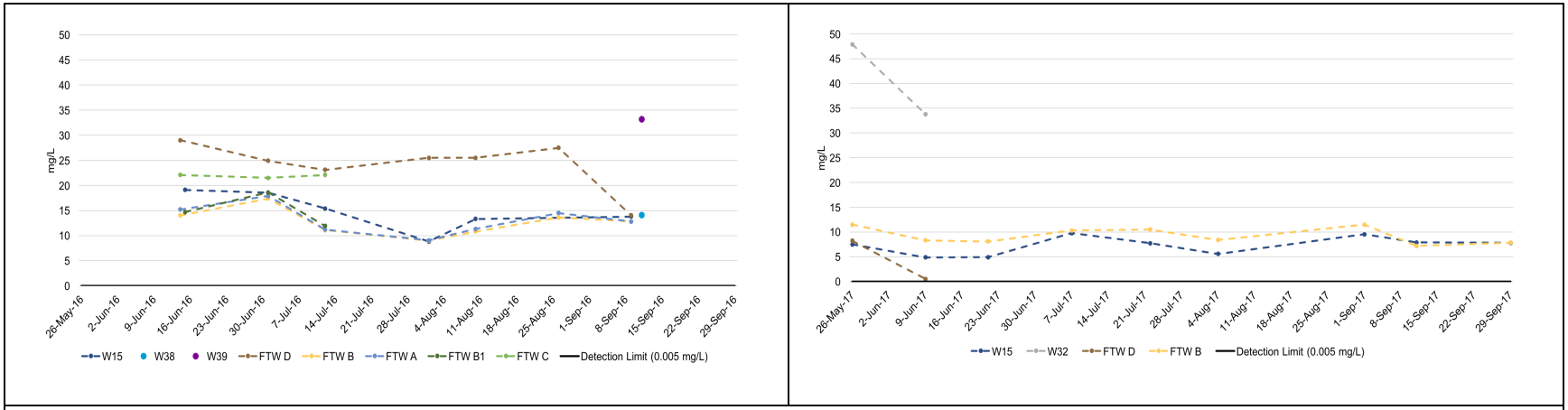


Figure 6 - Concentrations of nitrate at the W15 pond area.
The left graph shows 2016 data and the right graph shows 2017 data. W38 and W39 were sampled once in 2016 and are represented by single points.

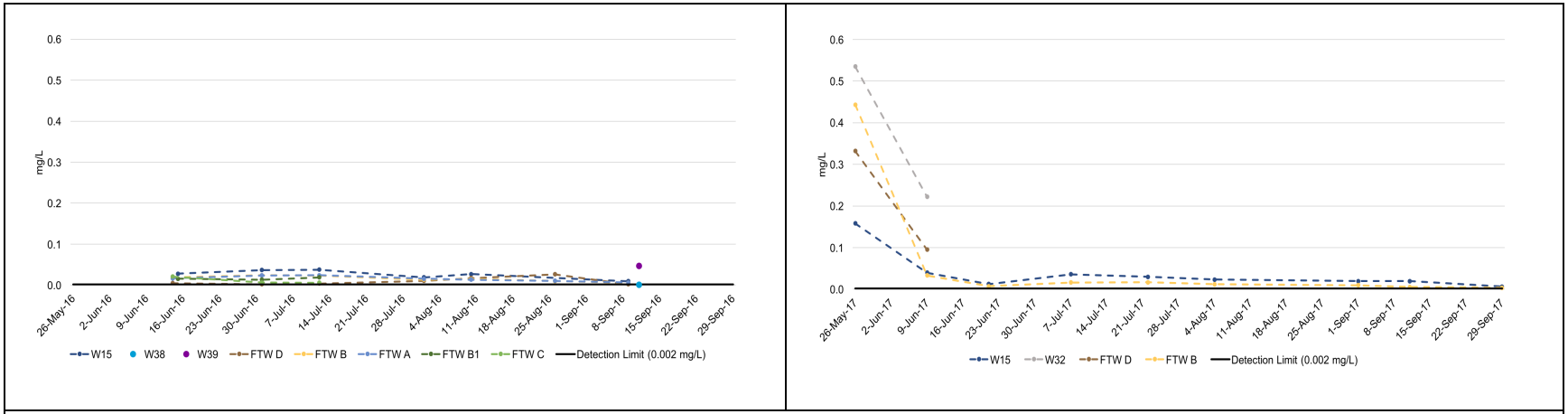


Figure 7 - Concentrations of nitrite at the W15 pond area.
The left graph shows 2016 data and the right graph shows 2017 data. W38 and W39 were sampled once in 2016 and are represented by single points.

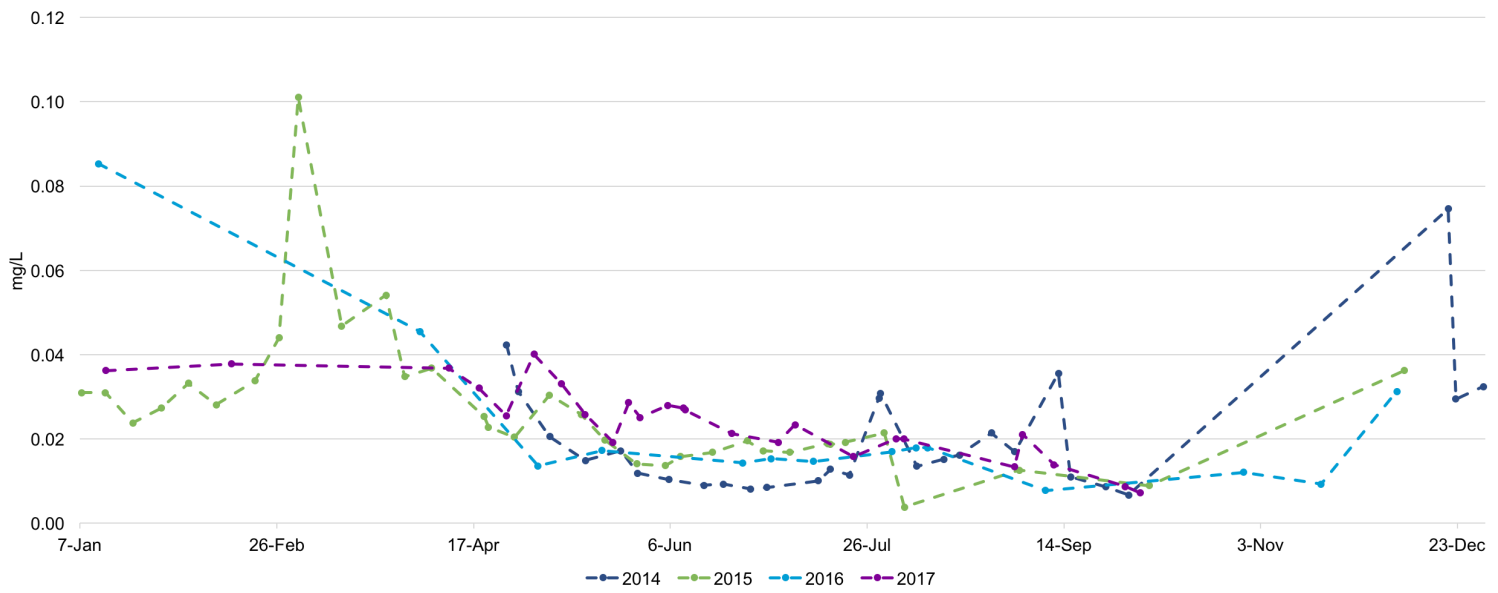


Figure 8 – Comparison of seasonal variation of dissolved copper at W15 pond for 2014 through 2017.
Dissolved copper concentrations from 2014 to 2017 at W15.

Table 4 – 2017 average, minimum, and maximum *in situ* water measurements.

Parameter		W15	W32	FTW-B	FTW-D
Temp (°C)	Min	2.4	0.9	0.5	1.9
	Avg	12.7	1.5	5.4	6.7
	Max	20.2	2.1	9.2	11.5
	SD	6.6	N/A	3.1	N/A
DO (mg/L)	Min	6.75	6.34	7.58	8.97
	Avg	9.36	9.72	10.12	10.75
	Max	13.66	13.09	12.91	12.53
	SD	2.16	N/A	1.60	N/A
SPC (µS/cm)	Min	456.2	886	529.3	529.3
	Avg	657.8	1107	712.7	555.8
	Max	852	1328	910	582.3
	SD	150.1	N/A	103.8	N/A
pH	Min	7.54	7.49	7.43	7.56
	Avg	7.72	7.57	7.60	7.66
	Max	7.87	7.65	7.77	7.75
	SD	0.11	N/A	0.13	N/A
ORP (mV)	Min	83.1	126.6	5.9	192.2
	Avg	132.0	195.6	94.3	210.6
	Max	206.1	264.5	205	229
	SD	46.5	N/A	73.7	N/A
# of samples	n	9	2	9	2

W32 and FTW-D have recorded values from 26-May-17 and 09-Jun-17.
W15 and FTW-B have recorded values from 26-May-17 to 29-Sep-17.
SD = standard deviation, N/A = SD was not calculated as there were not enough samples; DO = dissolved oxygen; SPC = specific conductance; ORP = oxidation-reduction potential.

4. Recommendations

The location of the conceptual FTW layout design proposed in 2015 should be reconsidered as the seeps contributing impacted water to the W15 area would not be intercepted by the current layout (since they are downstream). Therefore, treatment by terraced wetland is not recommended between W10 and FTW-B. The monitoring program in 2016 and 2017 has indicated that the creek is not the source of elevated metals. Rather, the source of constituents appears to be seeps reporting directly from the SWD to the west side of the W15 pond (i.e., W38 seep, via FTW-D monitoring point or subsurface flow). Additionally, since water from the creek (FTW-B) does not contribute to elevated constituents at W15, it will not require treatment.

Due to the construction of the ridgetop waste rock dump to the southeast of W15, the area should continue to be monitored and re-evaluated to determine if it is suitable for a treatment wetland or other passive treatment. It is recommended that the monitoring program continue in 2018. The 2018 spring and fall seepage monitoring program will identify if seep locations have migrated or if new seeps have emerged.

During the 2018 monitoring program, it is also recommended that the soil from the W15 pond be analyzed for leachable and total metals to determine if the elevated metal concentrations are being re-released from the soils and contributing to elevated concentrations in the W15 pond.

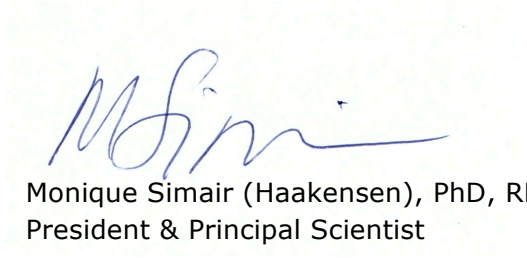
Once the chemistry and location of the seeps along the toe are delineated, flows should be assessed to determine potential load contributions to the W15 area and if treatment is feasible.

5. Closure

We trust the information herein satisfies your present requirements. Should you have any questions, please contact the persons listed below. We appreciate the opportunity to provide the services detailed in this report, and look forward to discussing any comments you may have.

Regards,

Contango Strategies Ltd.



Monique Simair (Haakensen), PhD, RPBio, PBiol, EP
President & Principal Scientist



Rachel Martz, BSc
Environmental Scientist



6. References

Contango Strategies Ltd. (2014). Minto Mine Constructed Wetland Treatment Research Program - Site Assessment. YESAB Registry documents number 2013-0100-256 and 2013-0100-257. March 24, 2014. Available at http://www.yesabregistry.ca/wfm/lamps/yesab/lowspeed/projectdetails.jsp;time=1510008082247.

Contango Strategies Ltd. (2015). Minto Mine Constructed Wetland Treatment Research Program – Demonstration Scale. March 22, 2015. Document #011_0315_01B.

Contango Strategies Ltd. (2016). Minto Mine Constructed Wetland Treatment Research Program – Demonstration Scale 2015 Update. March 2016. Document #011_0316_03B.

Contango Strategies Ltd. (2017a). Minto Mine Constructed Wetland Treatment Research Program – Demonstration Scale 2016 Update. February, 2017. Document #011_0217_05B.

Contango Strategies Ltd. (2017b). Future Terraced Wetland (FTW) Area Near W15 – Baseline Monitoring Program. Document #011_0217_06B.

Table A1 – FTW 2017 sampling program average, min, max concentrations for constituents of concern in mg/L.

Parameter	Results	Surface Water Monitoring Site	Seep Monitoring Sites	FTW Monitoring Site	
		W15	W32	FTW-B	FTW-D
Al-D	Min	0.0060	0.0020	0.0096	0.0035
	Avg	0.0167	0.0051	0.0191	0.0050
	Max	0.0364	0.0082	0.0363	0.0064
Al-T	Min	0.0553	0.0295	0.0237	0.2970
	Avg	0.3887	0.5148	0.1092	0.3135
	Max	1.6200	1.0000	0.3050	0.3300
Ag-D	Min	0.00001	0.00001	0.00001	0.00001
	Avg	0.00001	0.00001	0.00001	0.000012
	Max	0.00001	0.00001	0.00001	0.000013
Ag-T	Min	0.00001	0.00001	0.00001	0.00003
	Avg	0.00002	0.00002	0.00001	0.00003
	Max	0.00008	0.00004	0.00002	0.00004
Cd-D	Min	0.000005	0.000019	0.000005	0.000021
	Avg	0.000006	0.000039	0.000006	0.000033
	Max	0.000009	0.000059	0.000012	0.000046
Cd-T	Min	0.000005	0.000022	0.000005	0.000032
	Avg	0.000010	0.000053	0.000009	0.000039
	Max	0.000027	0.000084	0.000024	0.000047
Cr-D	Min	0.00010	0.00010	0.00016	0.00010
	Avg	0.00020	0.00010	0.00022	0.00010
	Max	0.00030	0.00010	0.00031	0.00010
Cr-T	Min	0.00020	0.00026	0.00022	0.00020
	Avg	0.00044	0.00833	0.00038	0.00039
	Max	0.00084	0.01640	0.00053	0.00058
Cu-D	Min	0.0088	0.0208	0.0037	0.0368
	Avg	0.0192	0.0293	0.0082	0.0449
	Max	0.0286	0.0377	0.0230	0.0529
Cu-T	Min	0.0194	0.0252	0.0046	0.0536
	Avg	0.0444	0.0617	0.0132	0.0695
	Max	0.1190	0.0982	0.0497	0.0853
Fe-D	Min	0.086	0.010	0.056	0.011
	Avg	0.132	0.013	0.098	0.014
	Max	0.202	0.016	0.159	0.016
Fe-T	Min	0.277	0.153	0.091	0.368
	Avg	0.809	0.662	0.353	0.420
	Max	3.180	1.170	1.340	0.472
Mo-D	Min	0.00194	0.00647	0.00193	0.00490
	Avg	0.00257	0.01214	0.00235	0.00696
	Max	0.00319	0.01780	0.00289	0.00902
Mo-T	Min	0.0022	0.0070	0.0023	0.0051
	Avg	0.0029	0.0122	0.0027	0.0070
	Max	0.0034	0.0173	0.0031	0.0088
Ni-D	Min	0.00059	0.00050	0.00054	0.00119
	Avg	0.00097	0.00050	0.00095	0.00119
	Max	0.00120	0.00050	0.00112	0.00119

Parameter	Results	Surface Water Monitoring Site	Seep Monitoring Sites	FTW Monitoring Site	
		W15	W32	FTW-B	FTW-D
Ni-T	Min	0.00094	0.00050	0.00090	0.00138
	Avg	0.00114	0.00059	0.00108	0.00152
	Max	0.00142	0.00067	0.00129	0.00166
Pb-D	Min	0.000050	0.000050	0.000050	0.000050
	Avg	0.000056	0.000050	0.000050	0.000050
	Max	0.000100	0.000050	0.000050	0.000050
Pb-T	Min	0.000050	0.000050	0.000050	0.000112
	Avg	0.000138	0.000243	0.000069	0.000134
	Max	0.000473	0.000436	0.000124	0.000155
Se-D	Min	0.0013	0.0125	0.0014	0.0007
	Avg	0.0018	0.0159	0.0019	0.0018
	Max	0.0028	0.0193	0.0023	0.0029
Se-T	Min	0.0012	0.0119	0.0015	0.0007
	Avg	0.0018	0.0153	0.0021	0.0017
	Max	0.0026	0.0186	0.0039	0.0027
Zn-D	Min	0.0010	0.0012	0.0010	0.0012
	Avg	0.0010	0.0016	0.0011	0.0026
	Max	0.0010	0.0020	0.0016	0.0039
Zn-T	Min	0.0030	0.0067	0.0030	0.0030
	Avg	0.0047	0.0073	0.0030	0.0046
	Max	0.0123	0.0078	0.0031	0.0062
Ammonia	Min	0.0155	0.0526	0.007	0.133
	Avg	0.0451	0.0656	0.026	0.245
	Max	0.0956	0.0786	0.054	0.356
Nitrate	Min	4.9	33.8	7.2	0.5
	Avg	7.3	40.9	9.3	4.4
	Max	9.8	47.9	11.5	8.2
Nitrite	Min	0.0062	0.2220	0.0036	0.0950
	Avg	0.0382	0.3785	0.0607	0.2135
	Max	0.1580	0.5350	0.4430	0.3320

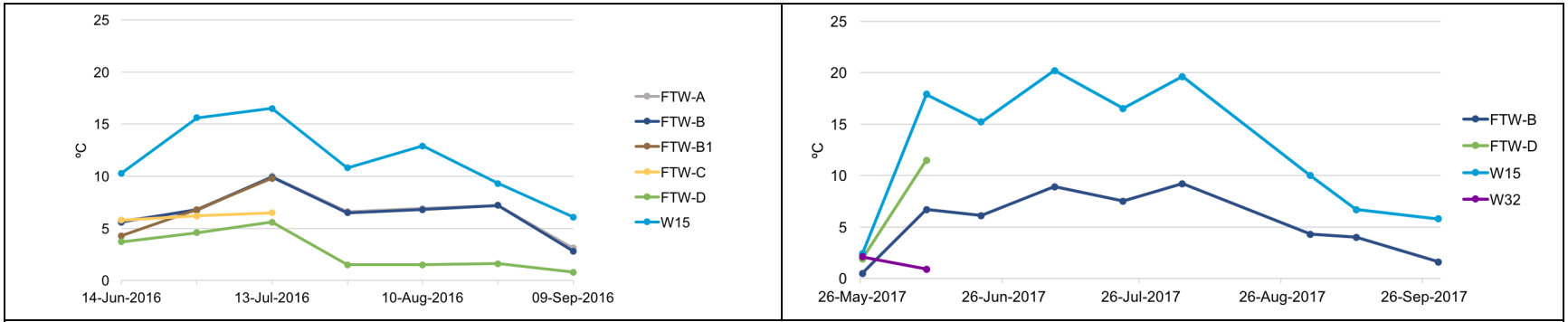


Figure A1 – Field temperature measurements.

The left graph shows 2016 data and the right graph shows 2017 data. Field measurements were not available for W32 in 2016. In 2016, FTW-C and FTW-B1 were only sampled three times as they were dry the for the remaining sampling dates. In 2017, FTW-D and W32 were only sampled twice as they were dry for the remaining sampling dates.

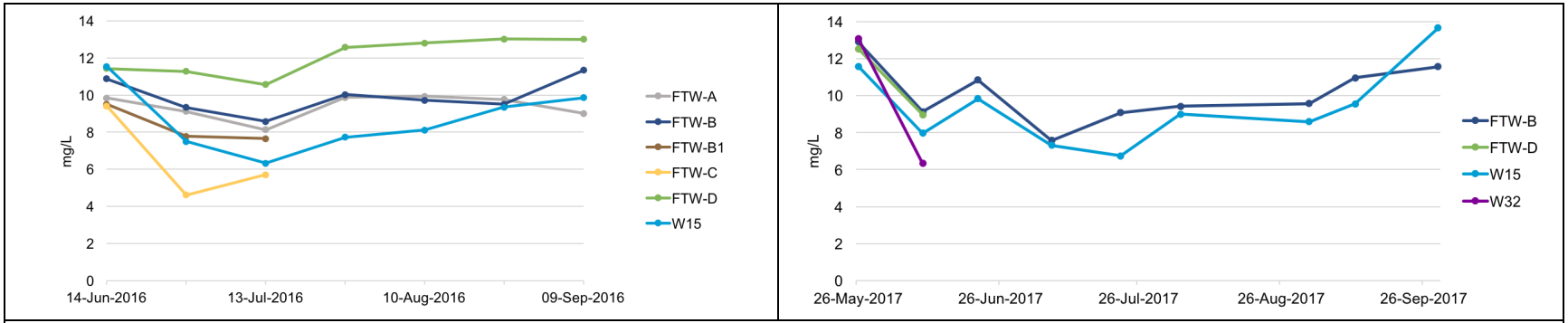
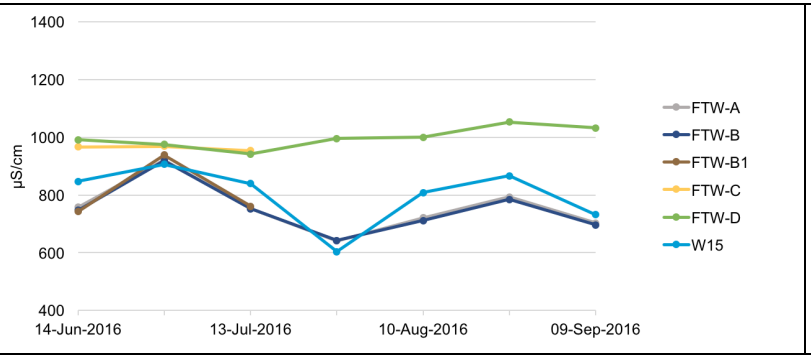


Figure A2 – Field dissolved oxygen (DO) measurements.

The left graph shows 2016 data and the right graph shows 2017 data. Field measurements were not available for W32 in 2016. In 2016, FTW-C and FTW-B1 were only sampled three times as they were dry the for the remaining sampling dates. In 2017, FTW-D and W32 were only sampled twice as they were dry for the remaining sampling dates.



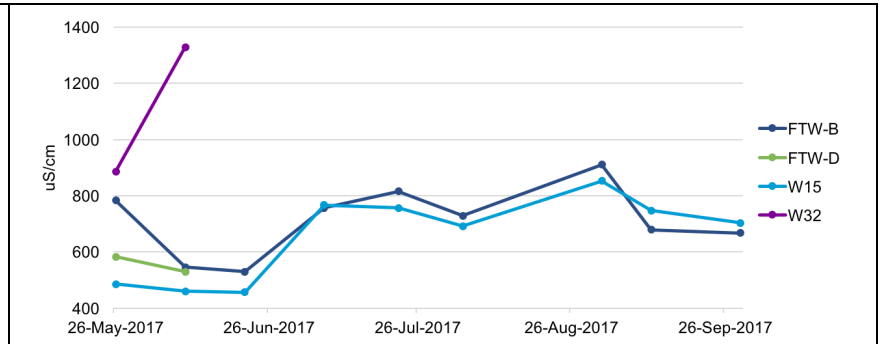
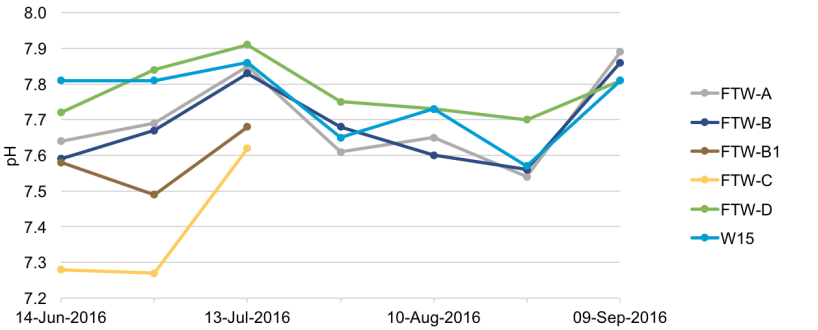


Figure A3 – Field specific conductance (SPC) measurements.

The left graph shows 2016 data and the right graph shows 2017 data. Field measurements were not available for W32 in 2016. In 2016, FTW-C and FTW-B1 were only sampled three times as they were dry the for the remaining sampling dates. In 2017, FTW-D and W32 were only sampled twice as they were dry for the remaining sampling dates.



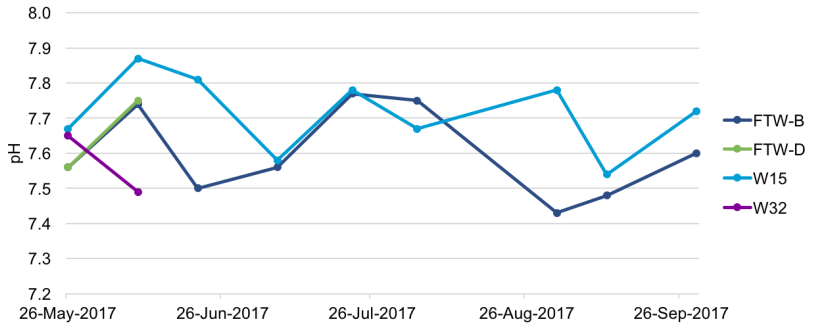
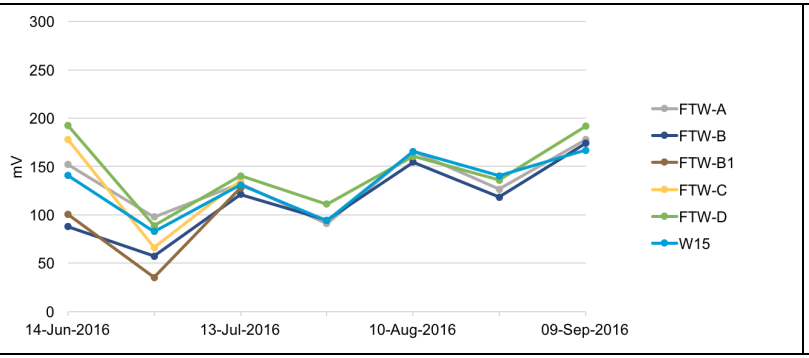


Figure A4 – Field pH measurements.

The left graph shows 2016 data and the right graph shows 2017 data. Field measurements were not available for W32 in 2016. In 2016, FTW-C and FTW-B1 were only sampled three times as they were dry the for the remaining sampling dates. In 2017, FTW-D and W32 were only sampled twice as they were dry for the remaining sampling dates.



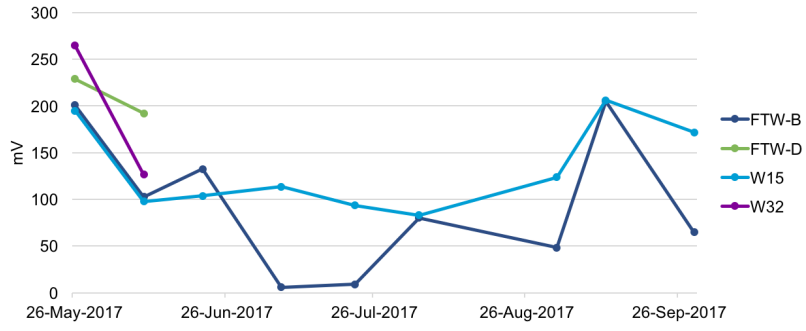


Figure A5 – Field oxidation reduction potential (ORP) measurements.

The left graph shows 2016 data and the right graph shows 2017 data. Field measurements were not available for W32 in 2016. In 2016, FTW-C and FTW-B1 were only sampled three times as they were dry the for the remaining sampling dates. In 2017, FTW-D and W32 were only sampled twice as they were dry for the remaining sampling dates.

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
2011				
05/05/2011 (Meeting) Whitehorse	Selkirk Development Corporation (SDC).	Contacts, roles and responsibilities	A discussion was held with members of SDC to provide updates on contacts within SFN and Minto.	Clarification of roles and responsibilities, exchange of contact information, areas identified for future collaboration.
27/05/2011 (Meeting) Whitehorse	Minto technical working group	Mine updates, review of current technical plans, indications for future plans given	This working group intended as a forum for input into conceptual technical plans from wide regulator/stakeholder group (YG, federal government, NGOs, etc.)	Information exchanged, input gathered on technical issues, plans for future licensing / permitting provided.
12/06/2011 (Meeting) Whitehorse	Minto Landing working group	Various technical and legal issues	Discussion between Minto on-site staff and SFN Lands department representatives regarding operation of the mine.	Communication intended to ensure up-to-date information being shared on daily activities at site.
16/06/2011 (Meeting) Whitehorse	BTWG	Geotechnical issues at Minto	Minto technical team prepared presentation to respond to questions from SFN's technical reviewers, mostly related to geotechnical issues related to Mill Valley Fill (MVF).	Discussion regarding various geotechnical aspects of mine facilities. Suggestions were made, recorded and plans made for further discussion in July/August.
30/06/2011 (Meeting) Pelly Crossing	Minto Landing working group	Various technical and legal issues	Discussion between Minto on-site staff and SFN Lands department representatives regarding operation of the mine.	Communication intended to ensure up-to-date information being shared on daily activities at site.
19/07/2011 (Meeting) Whitehorse	BWTG	Geotechnical issues at Minto	Geotechnical and water licence concerns, mostly related to MVF.	Discussion and information exchange related to issues, input gathered and plans made for future discussions.
17/08/2011 (Meeting) Pelly Crossing	Chief and Council	General updates	Regular meeting between respective leaderships.	Sharing of information in accordance with co-op agreement.
20/08/2011 (Presentation) Pelly Crossing	Community	General Assembly (GA) presentation	Minto representatives were given the opportunity to make a presentation to the entire SFN community as part of their annual GA.	Presentation by Minto employees on mine operations, future plans and question/answer period.
28/08/2011		Site tour		Sharing of information with community leaders.

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
(Site Tour) Minto	Selkirk renewable resource council (SRRC)		Representatives from the SRRC spent the day on site touring and partaking in dialogue with senior staff members who answered questions and provided demonstrations.	
31/08/2011 (Meeting) Minto	Minto technical working group	Mine updates, review of current technical plans, indications for future plans given	This meeting was held at site in order to provide context for members of the group who aren't familiar with site.	Field review of issues discussed during previous meeting.
01/09/2011 (Meeting) Whitehorse	Tri-partite working group	Initial discussions to begin work of this group (YG, Minto and SFN)	A discussion was held to clarify roles and responsibilities within this group.	Clarifications made, plans for future discussion made.
16/09/2011 (Meeting) Whitehorse	Minto technical working group	Discussion of permitting issues and technical work plans at Minto	This meeting was help in follow up to a proposed tour which many invitees could not attend. The discussion included updates on Minto activities and technical presentations by Minto's team on geotechnical matters affecting the mine including wall stability and dry stack tailings.	Provided responses to many technical questions, received input on plans, issues.
29/09/2011 (Meeting) Whitehorse	Various government agencies	Discussion of permitting issues and upcoming water licensing application	This meeting was held to discuss the role of Yukon government in the upcoming licensing process and provide insights into the scope of amendments being requested by Minto.	Clarification of roles, information sharing with government agencies.
03/10/2011 (Meeting) Pelly Crossing	Selkirk Chief and Council	General updates	Regular meeting between respective leaderships.	Sharing of information in accordance with co-op agreement.
04/10/2011 (Meeting) Whitehorse	Tri-partite working group meeting	Minto socio-economic effects monitoring	Collaboration with SFN and YG on how to meet requirements from Phase IV decision document re: socio-economic effects monitoring, as well as Minto's requirement to put forward socio-economic assessment for next YESAB application.	Clarified roles and responsibilities and discussed scope of effects monitoring, information resources, confidentiality, methods for collection of information, expected outcomes, etc.
14/10/2011 (Meeting) Whitehorse	Tri-partite working group follow up discussion	Socio-economic effects monitoring	Discussion with SFN team member on Tri-partite action items and follow up.	Meeting commitments for information sharing and planning next steps.
20/10/2011				

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
(Meeting) Whitehorse	Bi-lateral working group and Minto Landing working group	Operational updates, information on the preparation of upcoming water licence amendment	Water and waste management and Minto – current and future in the context of amendment #7 reasons for decision (issued in October 2011).	Sharing of views on Water Board RFD and discussion of how to proceed on water and waste management practices.
25/10/2011 (Meeting) Whitehorse	Tri-partite working group meeting	Socio-economic effects monitoring	Discussion with working group.	Discussion on progress to date.
28/10/2011 (Meeting) Whitehorse	Bi-lateral working group meeting	Water Licence amendment #8 review with SFN	Discussion with technical team and community members.	Information sharing, receiving input for inclusion in Minto plans, answering questions from community members and technical team. Discussion extended to closure and reclamation issues.
04/11/2011 (Meeting) Pelly Crossing	SFN social programs coordinator	Minto participation/funding social programs	Discussion to understand how Minto can meet commitments to social well-being and ideas for the holidays.	Information sharing, ideas for Minto's involvement based on needs in the community.
07/11/2011 (Meeting) Pelly Crossing	Selkirk Development Corp. leadership	Current and future business opportunities	Minto site updates regarding contracts and discussion of upcoming opportunities.	Information sharing, updates on contact information.
14/12/2011 (Meeting) Pelly Crossing	Selkirk Chief and Council	General updates	Regular meeting between respective leaderships.	Sharing of information in accordance with Co Op Agreement.
2012				
23/01/2012 (Meeting) Pelly Crossing	Selkirk Development Corp. leadership	Current and future business opportunities	Minto site updates regarding contracts and discussion of upcoming opportunities.	Information sharing, updates on contact information.
03/02/2012 (Meeting) Whitehorse	Bi-lateral working group meeting	Ongoing water licensing issues	Water management issues related to Mill Valley Fill; discussed Minto response to water licence information request.	Information sharing, receiving input for inclusion in Minto plans, answering questions on Minto technical documents/submissions to Water Board. Discussion extended to closure and reclamation issues.
13/02/2012	Bi-lateral working group meeting			

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
(Meeting) Whitehorse		Water licence amendments and closure	Updates on Minto progress related to water licensing.	Input received, information shared, plans made for future engagement.
07/03/2012 (Meeting) Pelly Crossing	SFN employment liaison	Welcome meeting	Employment of SFN members at Minto mine.	Information sharing and action items related to increased hiring of SFN members at Minto Mine.
21/03/2012 (Site Tour) Minto	Community	Site tour	Open invite to community members to tour the site.	Information sharing, relationship building.
23/03/2012 (Meeting) Pelly Crossing	Bi-lateral working group	Water Licensing and upcoming YESAB submission	Water and waste management, response to information requests from the water board. Took questions from community on various Minto issues.	Information sharing, receiving input for inclusion in Minto plans, answering questions on Minto technical documents/submissions to Water Board.
03/04/2012 (Presentation) Pelly Crossing	Community	Employment	SFN Community hosted Minto representatives for a job fair style event. Presentations made providing more information on careers at Minto Mine.	Contacts made, information shared, questions answered.
24/04/2012 (Meeting) Whitehorse	Bi-lateral working group	Water licensing and upcoming YESAB assessment application	Water and waste management issues, information requests related to water licence amendment #8.	Ongoing updates provided, information shared, plans for future engagement made.
24/05/2012 (Meeting) Whitehorse	Bi-lateral working group	Water licensing and upcoming YESAB assessment application	Water and waste management issues, information requests related to water licence amendment #8.	Ongoing updates provided, information shared, plans for future engagement made.
31/05/2012 (Meeting) Pelly Crossing	SFN citizens	Socio-economic study	Two community meetings in Pelly Crossing (afternoon and evening). Over 20 community members attended the afternoon meeting and 10 attended the evening meeting.	Discuss and identify socio-economic values and potential effects.
07/06/2012 (Meeting) Pelly Crossing	Bi-lateral working group	Water licensing, upcoming amendment #8 hearing review	Water and waste management issues, information requests related to water licence amendment #8.	Ongoing updates provided, information shared, plans for future engagement made.

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
14/06/2012 (Meeting) Whitehorse	Minto technical working group	Water licence upcoming hearing related to amendment #8	Meeting with wide reviewer audience to explain technical information that will be discussed at upcoming water licence amendment hearing.	Minto updates provided, review of technical information submitted to Water Board.
15/06/2012 (Meeting) Whitehorse	Bi-lateral working group	Water licensing, upcoming amendment #8 hearing review	Water and waste management issues, information requests related to water licence amendment #8.	Ongoing updates provided, information shared, plans for future engagement made.
18/07/2012 (Site Visit) Minto Mine	Regulator	Closure	Site tour with YG, Energy, Mines, and Resources representatives.	Orientation to mine site and discussion of closure.
09/08/2012 (Newsletter) Pelly Crossing	SFN citizens	Socio-economic study	Newsletter sent to all SFN citizens and community members in Pelly Crossing.	Newsletter provided information on socio-economic data gathering and project activities.
14/08/2012 (Meeting) Pelly Crossing	SFN citizens	Socio-economic study	Community meeting facilitated by Klohn Crippen with participation of Minto representatives.	Informal discussion, with Power Point presentation, to discuss socio-economic values and potential effects as they relate to Phase V/VI.
2012/08/14-17 (Interviews) Pelly Crossing	SFN citizens/ community members	Socio-economic study	Thirteen individual interviews with community members and individuals working in the community.	Socio-economic data collection.
15/08/2012 (Presentation) Pelly Crossing	SFN Elders' Council	Socio-economic study	Presentation to Elders' Council (23 Elders and 1 SFN Coordinator).	Informal discussion, with PowerPoint presentation, to discuss socio-economic values and potential effects as they relate to Phase V/VI.
15/08/2012 (Meeting) Whitehorse	BTWG	Various technical issues	All-day meeting with approximately 5 technical representatives from each of Minto and SFN.	Structured technical discussions regarding: closure, permit applications (Phase V/VI), site activities, and status.
15/08/2012 (Meeting)	SFN Socio-economic working group representatives	Socio-economic study	Meeting (informal dinner) between Klohn Crippen and SFN socio-economic working group representatives.	Discussion of comments received from Socio-economic Tri-partite Working Group.

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
Whitehorse				
23/08/2012 (Telephone Interview)	SFN citizens/ community members	Socio-economic study	Individual interview with one community member.	Socio-economic data collection.
28/09/2012 (Letter)	Chief McGinty	Phase V/VI	Letter sent to Chief McGinty.	Written request to discuss details of Phase V/VI with Chief and Council as well as with the community.
16/10/2012 (Meeting) Pelly Crossing	SFN citizens	Introduction to Phase V/VI	Afternoon and evening meeting (with dinner) for SFN citizens and community members. Approximately 40 people attended. Translator (Northern Tutchone) provided.	Informal discussion, with PowerPoint presentation and maps, of activities to be undertaken for Minto Phase V/VI expansion. General Q&A.
16/10/2012 (Interviews) Pelly Crossing	SFN citizens/ community members	Socio-economic study	Two individual interviews with community members and individuals working in the community.	Socio-economic data collection.
2012/10/22-26 (Interviews) Pelly Crossing	SFN citizens/ community members	Socio-economic study	Sixteen individual interviews with SFN citizens and community members.	Gathering feedback on identified socio-economic values.
22/10/2012 (Meeting) Pelly Crossing	Klohn Crippen and SFN's socio-economic working group representatives	Socio-economic monitoring program	Meeting with Klohn Crippen and SFN SE working group representatives.	Discussion of socio-economic baseline data being collected by Klohn Crippen.
29/10/2012 (Meeting) Pelly Crossing	Chief and Council	Various	Attended SFN Chief and Council meeting.	Discussed BTWG, mine tours, socio-economic work, consultation, river crossing, water quality monitoring proposal, training opportunities for employees.
2012/10 (Newsletter)	SFN citizens/ community members	General mine information and Phase V/VI information	Newsletter sent to all SFN citizens and community members.	Information provided on mine team, jobs, training, Phase V/VI, opportunities for feedback.
2012/10 (Newsletter)	SFN citizens/ community members	Socio-economic study newsletter	Newsletter sent to all SFN citizens and community members.	Provided a summary of study activities and a questionnaire on valued socio-economic components.
01/11/2012 (Meeting)	EMR and SFN	Closure	Meeting with YG Energy, Mines and Resources and SFN.	Closure cost update for Phase IV.

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
Whitehorse				
08/11/2012 (Meeting) Whitehorse	BTWG	Waste management	Approximately a half dozen technical representatives from each of Minto and SFN.	Members from both Minto's and SFN's technical teams met to discuss the creation of technical working groups as well as Minto providing an overview of Phase V/VI.
09/11/2012 (Call)	SFN technical consultant(s)	Closure	Call between Minto and SFN technical consultants.	Discussions about planning process, MAA and FMEA for closure.
14/11/2012 (Call)	SFN technical consultant(s)	Closure	Call between Minto and SFN technical consultants.	Discussion about closure planning process.
15/11/2012 (Call)	SFN technical consultant(s)	Waste management and closure planning	Call between Minto and SFN technical consultants.	Discussions about waste management and closure planning.
16/11/2012 (Call)	SFN technical consultant(s)	Phase V/VI	Call between Minto and SFN technical consultants.	Discussion about Phase V/VI, 2012 hydrological program, potential waste disposal options.
20/11/2012 (Presentation) Whitehorse	Public	Minto Mine	Presentation at Yukon geoscience forum.	Provided information regarding Minto Mine operations and plans.
20/11/2012 (Call)	SFN technical consultant(s)	Waste management	Call between Minto and SFN technical consultants.	Alternatives assessment process, waste rock management.
2012/11/26-28 (Mine Tour) British Columbia	SFN Elders and leadership	Closure	Minto and technical representatives, along with SFN Leadership and Lands department representatives visited Highland Valley Copper Mine.	Focus of the mine tour was on closure; successes and challenge of both standard and innovative measures.
29/11/2012 (Meeting) Whitehorse	Regulators, SFN	Introduction to Phase V/VI	Half day meeting with various federal and territorial regulators including: EMR, Water Board, SFN, YESAB, Environment Canada, YG Water Resources.	Informal discussion, with PowerPoint presentation, of activities to be undertaken for Minto Phase V/VI expansion. General Q&A.
03/12/2012 (Call)	SFN technical consultant(s)	Waste management	Call between Minto and SFN technical consultants.	Alternatives assessment process, Phase V/VI, water management.
09/12/2012	SFN technical consultant(s)	Phase V/VI		

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
(Call)			Call between Minto and SFN technical consultants.	Review of BTWG meeting outcomes, closure planning.
11/12/2012 (Meeting) Whitehorse	EMR and SFN	Closure	Meeting with YG Energy, Mines and Resources and SFN.	Review of new draft ERM reclamation closure plan guidelines.
13/12/2012 (Workshop) Whitehorse	BTWG	Waste management alternatives assessment	One-day workshop. Approximately 8 representatives from each of SFN and Minto.	Identification of preferred waste management option.
18/12/2012 (Meeting) Whitehorse	SFN leadership representatives	Consultation and community engagement	Meeting with Capstone leadership and SFN leadership representatives.	Discussion of long-term scheduling and Leadership expectations regarding community engagement.
19/12/2012 (Call)	YESAB	Update	Call with Katrine Frese (Mayo DO) and Bengt Pettersson (Exec Director) from YESAB with Minto representatives.	Overview/history of project and upcoming proposal for Phase V/VI.
2013				
26/01/2013 (Meeting) Whitehorse	SFN citizens	Introduction to Phase V/VI	Public meeting for SFN citizens living in Whitehorse (approximately 6 SFN citizens attended).	Informal discussion, with PowerPoint presentation and maps, of activities to be undertaken for Minto Phase V/VI expansion. General Q&A.
08/02/2013 (Call)	SFN's socio-economic representatives	Traditional knowledge and socio-economics	Call with Minto representatives and Klohn Crippen with SFN's socio-economic working group representatives.	Status and progress update, Q&A, traditional knowledge, socio-economic study.
12/02/2013 (Call)	SFN technical consultant(s)	Closure	Call between Minto and SFN technical consultants.	FMEA roll-up, closure planning.
18/02/2013 (Site Visit) Minto Mine	YESAB	Orientation	Site tour with YESAB Executive Director, Mayo DO Assessment Officer and DO Support.	Tour of the mine site.
19/02/2013 Call	SFN leadership representative	Water Board filings and technical details	Call between Capstone Leadership and SFN Leadership rep.	Discussion of technical issues, level of detail for WB filings.

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
10/03/2013 Call	SFN leadership representative	Closure	Call between Capstone Leadership and SFN Leadership rep.	Discussion regarding closure bonding.
14/03/2013 (Site Visit) Minto Mine	YESAB	Orientation	Site tour with Mayo DO Manager and Teslin DO Manager.	Tour of the mine site.
22/03/2013 (Meeting) Whitehorse	Tri-partite working group	Socio-economic effects monitoring	Meeting of the socio-economic Tri-partite working group (Minto, SFN representatives and YG).	Discuss components of a socio-economic monitoring program.
25/03/2013 (Call)	SFN Socio-economic representatives	Socio-economic baseline report	Call with Minto and rep, SFN representatives, and Klohn Crippen.	Provide comments and feedback on the draft socio-economic baseline report.
05/04/2013 (Open House) Pelly Crossing	SFN citizens and community members	Employment opportunities	Presentation at an Open House in the Lync Building.	Presentation regarding training and employment opportunities.
04/04/2013 (Call)	SFN leadership representative	YESAA application	Call between Capstone Leadership and SFN leadership rep.	Discussion about project proposal and potential meeting with Chief and Council.
25/04/2013 (Meeting) Whitehorse	Tri-partite working group	Socio-economic effects monitoring	Meeting of the socio-economic Tri-partite working group (SFN, Minto, and YG).	Discus components of a socio-economic monitoring program.
30/04/2013 (Meeting) Whitehorse	BTWG	Phase V/VI Project	Meeting of the Bi-lateral working group (SFN, Minto) and technical representatives.	Discuss various components of the Phase V/VI project and the project proposal.
24/06/2013 (meeting) Whitehorse	YWB	Lessons learned from phase IV amendment process	Meeting between permitting manager and YWB secretariat to discuss lessons to be learned from the previous water licensing process.	
2013/08/08 (site tour) Minto Mine	YWB	Site tour for YWB Board members and staff	Tour for Board members and YWB staff to familiarize themselves with the project.	Presentations and field visit

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
13/09/2013 (meeting) Whitehorse	EMR	Phase IV RCP review process and status of phase V/VI proposal	Discussion with EMR staff on the process for reviewing the phase IV RCP. Minto concerned about significant overlap between EMR process and YWB process. Discussion about timelines of YESAB process and preparation of licence applications. Minto concerned about impact from W2 water standards.	Minto to explore options for conducting some of phase V/VI work prior to receiving a water licences,
10/10/2013 (meeting) Whitehorse	YESAB	Clarification on information request	Minto required clarification on three questions in information request (B).	Clarification received
17/10/2013 (meeting) Whitehorse	SFN leadership	Principals meeting between Minto and Selkirk	Discussion between Selkirk Chief and council and Minto senior management including water quality objectives at closure	Commitment to develop WQO collaboratively.
23/10/2013 (meeting) Whitehorse	EMR	Permitting process update and QML application logistics	Minto provided status update on YESAB process and application development. Discussion on format of QML application.	
28/10/2013 (meeting) Vancouver	BTWG	Water quality objectives	Technical discussion on post closure water quality objectives.	
31/10/2013 (meeting) Whitehorse	EMR	Follow up meeting to 2013/10/23 meeting on application logistics	Staff level discussion on details of application format.	
06/11/2013 (Community	SFN citizens	Community meeting to update SFN members about	Presentations, question and answer session, discussion on various topics including scope of phase V/VI proposal and water quality	Information meeting

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
meeting) Pelly Crossing		Minto Mine, including permitting and closure planning	objectives at closure.	
25/11/2013 (meeting)	YESAB	Clarification remainder of YESAB process and White River First Nation participation	Discussion on remaining steps in YESAB process and related information needs. YESAB provided clarification on how White River First Nation claims will be incorporated into the process.	
28/11/2013 (meeting) Whitehorse	EMR	Closure cost discussion	Discussion between Minto, EMR and respective consultants on closure cost estimates in relation to the phase IV RCP and the resulting security bond.	Minto to revise cost estimates following input from EMR
2014				
13/01/2014 (meeting) Whitehorse	YWB	Clarification on guidance document	Discussion to obtain clarity on information requirements under YWB's various guidance documents.	
2014/02/07 (call) Whitehorse/ Vancouver	EMR	Legal questions around YESAB process	Discussion on definition of a Decision Body under YESAA and related mandate questions.	Increased clarity on decision bodies.
13/02/2014 (meeting) Whitehorse	YESAB	Main Dam and RCP information requirements	Discussion on the information requirements for the Main Dam design and the closure plan; where does environmental assessment end and water licensing begin?	Increased clarity on the process, what information is needed and what information can re reasonably provided.
14/02/2014 (meeting)	BTWG	Water Quality Objectives	Full day workshop on water quality objectives during operations and post closure. In depth discussion on methods to derive objectives, the application of objectives, and the feasibility of achieving certain	Agreement on possible approach to setting water quality objectives.

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
Whitehorse			objectives	
10/03/2014 (meeting) Whitehorse	EMR	Phase IV closure costing	Follow up discussions from 2013/11/28 to further refine closure cost estimates and resulting security bond	
30/04/2014 (meeting) Whitehorse	YWB	Licence renewal issues	Discussion to clarify the requirements and the limitations of water licence renewal application if renewal and phase V/VI are to be combined in one process.	Increased process clarity.
02/05/2014 (meeting) Whitehorse	EMR	Evaluation Report	Discussion on how to interpret conditions in the YESAB evaluation report and how they may be translated into conditions in a decision document	Minto concerned about the nature and timing of many of the conditions. Some raise mandates issues, e.g. can YESAB create a de-facto veto right for a stakeholder?
13/06/2014 (meeting) Whitehorse	BTWG	Decision Document and water quality objectives	Discussion on timing of licence applications. Discussion on Decision Document and resulting licence application requirements.	Agreement on a tentative approach for reaching consensus on as many items as possible prior to water board hearings.
20/06/2014 (meeting) Whitehorse	BTWG	Water quality objectives	In depth discussion of water quality objectives and approach to determining them. .	Agreement on general approach, further work required on defining contaminants of concern.
09/07/2014 (Meeting) Minto	SFN Leadership	Phase V/VI	Site tour and discussion regarding Phase V/VI	Concerns from SFN shared with Minto
14/07/2014 (Meeting)	BTWG	Phase V/VI	COPC selection criteria, proposed WQOs, proposed effluent quality limits and revised predictions	

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
Whitehorse				
16/07/2014 (Meeting) Pelly Crossing	SFN Leadership	Chief and Council Meeting	Regular Minto and Chief and Council Meetings	
18/07/2014 (Meeting) Pelly Crossing	Yukon Government, Environment Canada, SFN	Ground Water Modeling Workshop	Ground Water Modeling Workshop	
21/07/2014 (Meeting) Whitehorse	Tripartite	Project plan and desired critical path	Project scope, process, mine development options, preview licence application	
29/07/2014 (Meeting) Whitehorse	Regulators, SFN	Stakeholder Engagement	To facilitate stockholders' upcoming review of Minto's Phase V/VI licence applications	
22/08/2014 (Meeting) Whitehorse	EMR and YWB	Licence application	To facilitate stockholders' upcoming review of Minto's Phase V/VI licence applications	
31/07/2014 (Meeting) Whitehorse	Regulators, SFN	Stakeholder Engagement	To facilitate stockholders' upcoming review of Minto's Phase V/VI licence applications	
16/09/2014 (Meeting) Whitehorse	SFN	Water quality objectives	Discussions regarding water quality objectives for the Phase V/VI applications	
18/09/2014 (Meeting) Whitehorse	EMR	General Updates	Discussing operational updates	
18/09/2014	SFN	3rd party geotechnical review	Discussing results of the geotechnical review	

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
(Meeting) Whitehorse				
08/10/2014 (Meeting) Minto	SFN, EMR	Site Visit	Closure planning site visit	
11/11/2014 (Meeting) Whitehorse	SFN, EMR	Closure Planning	Discussing the details of the submitted closure plan	
2015				
08/01/2015 (Meeting) Whitehorse	BTWG	Water Quality Objectives	Discussions regarding water quality objectives for the Phase V/VI applications	
20/08/2015 (Meeting) Whitehorse	BTWG	Reset Meeting	Discussion around engagement opportunities for SFN regarding the RCP	Agreed to focus on the main points such as closure WQO, design criteria and closure objectives
10/09/2015 (Meeting) Minto	Yukon Government	Site Tour	Site tour for Yukon Environment personnel.	
15/09/2015 (Meeting) Minto	BTWG	Site Tour	Site tour for SFN technical staff.	Discussion regarding closure engagement opportunities and applying for the extension
05/11/2015 (Meeting) Phone	BTWG	Water quality objectives	WQO framework discussion with side discussion regarding AMP, Cu Toxicity Testing and Water Management review and approval process	Beginning of discussions on the WQO framework for the closure plan.
06/11/2015 (Meeting)	BTWG	Water quality objectives	WQO framework discussion with side discussion regarding AMP, Cu Toxicity Testing	Beginning of discussions on the WQO framework for the closure plan.

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
Phone			and Water Management review and approval process	
16/11/2015 (Meeting) Whitehorse	Tripartite	Extension request	Discuss regarding the scope of the upcoming RCP and the proposed extension request.	EMR and SFN to support the extension request.
11/12/2015 (Meeting) Whitehorse	BTWG	Water Quality Objectives	Discussion regarding the COPC, background data set, mock WQO using the formula in the WUL. Discussion regarding attainment.	SFN to provide an attainment document and justification for inclusion of hardness in the WQO.
2016				
11/01/2016 (Meeting) Phone	BTWG	Primary Water Conveyance	Discussion regarding primary water conveyance alignments and options as well as design criteria and potential locations for wetlands	General agreement on water conveyance alignment.
23/02/2016 (Meeting) Phone	SFN Leadership	Chief and Council Meeting	General site update	
01/03/2016 (Meeting) Whitehorse	BTWG	Wetland Options	Discussions regarding different wetland configurations and options. Execution strategy.	SFN would like to see a predictable system.
05/04/2016 (Meeting) Phone	BTWG	Closure Conveyance Routing	Discussed the impact of routing water through the Area 2 Pit vs routing around Area 2 pit on the downstream hydrology. This was to address concerns raised previously by SFN.	It was agreed that there is little difference between routing on the downstream hydrology. SFN raised another concern on if area derived dilutions ratios are appropriate.
03/05/2016 (Meeting) Phone	BTWG	Wetland Concept	Discussions regarding wetland concept for WSP area and water conveyance in the area	Overall consensus on the wetland configuration
11/05/2016 (Meeting) Whitehorse	Tripartite	RCP Progress Update, Environment Audit	General RCP progress update and discussions regarding the environment audit proposal.	SFN will not support SRK carrying out the audit.

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
19/05/2016 (Meeting) Phone	BTWG	Water Quality Objectives	Discussions regarding the final step of the WQO framework and how to move forward	Decision to get together and work through how to use the background water quality dataset.
02/06/2016 (Meeting) Vancouver	BTWG	Water Quality Objectives	Worked through how to use the background water quality dataset.	An agreed upon process for determining background water, water quality dataset.
14/06/2016 (Meeting) Pelly Crossing	SFN	SFN Engagement	Discussions around SFN engagement on multiple plans and programs including the RCP, Vegetation Metal Uptake Program and Environmental Audit.	General update
29/06/2016 (Meeting) Whitehorse	YWB	EMSRP and MVFES2 information request clarification	Discussed technical details regarding the MVFES2 Expected Performance and Evaluation Criteria Report and EMSRP.	Clarification received
06/07/2016 (Meeting) Minto	Tripartite	Closure concepts	Tour for EMR, YG Env. and SFN	For parties to site and advancement of operations
13/07/2016 (Meeting) Phone	BTWG	EMSRP and Groundwater Monitoring Plan	Discussion regarding the location of proposed wells. Discussed options and preferred conceptual plan for the lysimeters.	Overall general agreement with proposed well locations. General consensus on developing lysimeters that could be used to test closure cover performance.
14/07/2016 (Meeting) Phone	BTWG	Vegetation Metal Uptake Program	SFN technical advisor provided feedback on the acceptance of the proposed vegetation species to be used in the sampling program.	SFN technical advisor agreed to the proposed approach submitted to SFN for consideration.
15/07/2016 (Meeting) Phone	BTWG	W2 Background water quality determination approach	Reviewed the work by Minnow with SFN technical advisors regarding the development of the background water quality approach.	General consensus that the approach acceptable.
9/09/2016 (Meeting)	BTWG	Advancement of closure WQO development	Comparison of proposed WQO to WQ prediction.	SFN technical team to discuss with SFN leadership.

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
Whitehorse				
4/10/2016 (Meeting) Whitehorse	YG	Closure costing site tour	General site tour with a focus on reclamation closure measures	
3/11/2016 (Meeting) Whitehorse	BTWG	Discussions regarding closure water quality objectives.	Agreement on numerical closure WQO non degradation and 50% of assimilative capacity/ closure monitoring frequency and closure AMP also discussed	Agreement on numerical values but work remains on attainment criteria and closure AMP.
24/11/2016 (Meeting) Whitehorse	Tripartite	SFN and EMR RCP Assessment determination and Area 2 Stage 3 pit concerns	Discussed EMRs and SFNs assessment determination regarding the RCP application to the YWB. Also discussed SFN technical review comments regarding the Area 2 Stage 3 pit Mine Development and Operation Pit	Provide stability of new MPD design, water balance for Area 2 Pit a figure showing SAT dump.
5/12/2016 (Meeting) Phone	BTWG	Minto response to YWB IR	Discussion regarding the Minto's planned response to the YWB information request.	
9/12/2016 (Meeting) Pelly Crossing	SFN Leadership	Principals Meeting	General site update	
13/12/2016 (Meeting) Whitehorse	YG	YG AMP Review comments	Discussed YG AMP review comments and Minto concerns with those requests.	Minto clarified issued already addressed.
15/12/2016 (Meeting) Phone	BTWG	Closure WQO attainment criteria and Closure AMP	Discussed how closure WQO would be applied and how it would work within the Closure AMP.	No agreement was reached on the final WQO attainment or details of Closure AMP. Minto considered making some changes to the Closure AMP.

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
2017				
27/02/2017 (Emails) Emails	BTWG	Review of closure WQO section in the RCP	Review of closure WQO section in the RCP and looking for SFN feedback.	Feedback received from SFN technical consultants agreeing with the content of the information.
8/03/2017 (Meeting) Pelly Crossing	Pelly Community Members	General update	Review of 2017 mine plan, results of 2016, employment opportunities, RCP overview.	Questions period – Jobs, business opportunity, closure questions.
27/04/2017 (Meeting) Whitehorse	BTWG	Closure of Minto Waste Rock Dumps	Engagement and getting feedback on the implementation of progressive reclamation	SFN technical group provided feedback in person only and did not follow up with written comments
23/05/2017 (Meeting) Whitehorse	YG	RCP Costing review	Discussion regarding YG consultant costing review.	
10/07/2017 (Tour) Minto Mine	SFN Leadership	General site tour	Site tour with a focus on progressive reclamation	
13/07/2017 (Meeting) Pelly Crossing	SFN Leadership	Principals meeting	General site update	
18/07/2017 (Meeting) Whitehorse	YWB	YWB licence amendment	Discussed process for submitting the YWB licence amendment	
10/10/2017 (Tour) Minto Site	YG and YWB	General site tour	Discussion regarding groundwater.	

Date, Event, Location	Stakeholder/Group	Issue	Engagement Details	Outcome/Purpose
26/10/2017 (Meeting) Whitehorse	YG	Groundwater	Discussed water in pits and groundwater. Brief discussions regarding operational AMP changes.	
17/11/2017 (Meeting) Phone	BTWG	Bioassay frequency reduction	Minto proposed a reduction in bioassay monitoring frequency for the upcoming licence amendment application	Minto to draft up a memo outlining changes.
15/12/2017 (Meeting) Whitehorse	BTWG	General application and RCP update	Discussed changes to the closure AMP, FMEA, WQ predictions, and MWD wrap	Another meeting to be held with SFN technical team providing feedback on the Closure AMP and Wetland treatment systems.
2018				
9/01/2018 (Meeting) Whitehorse	BTWG	Closure AMP and Wetland Treatment System Review	Discussed SFN feedback on Closure AMP and Lorax Wetland review comments.	SFN technical team to provide written Closure AMP feedback.

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Memorandum

To: Ryan Herbert (Minto Explorations Ltd.)

From: Jim Theriault, Jim Harrington, Scott Keesey (Alexco Environmental Group, Ltd.)
Monique Haakensen, (Contango Strategies Ltd.)

Date: February 26, 2016

Re: **Re-evaluation of Candidate Passive Treatment Technologies for Minto Mine Closure**

1 INTRODUCTION

Passive treatment of mining impacted waters (MIW) in closure and post-closure is being evaluated for the Minto Mine. The Reclamation and Closure Plan (RCP) for the site proposes a transition from active to passive water treatment, and research has been underway to refine performance expectations for the technologies most likely to be successful at the site. These include constructed wetland treatment systems, bioreactors, and in-pit batch biological treatment of MIW. These technologies were previously selected in connection with preparation of RCP v4.0 (Phase IV Mine Plan), based on a methodical evaluation of appropriate technologies for the site.

This memo reviews the decision-making process that was employed to identify the preferred passive treatment technologies and also presents the results of a recent re-evaluation of potential passive treatment technologies that could be effectively incorporated into the new Phase V/VI Minto RCP. It also documents the technologies and best practices that have been incorporated into the reclamation design that ultimately have resulted in the water quality that can be treated by the passive or semi-passive technologies proposed. This includes a number of types of mitigation measures which prevent constituents from becoming part of the MIW, which ultimately results in a minimization of treatment requirements. These prevention technologies are listed in Table 1, and include:

- Administrative controls,
- Backfilling and subaqueous disposal,
- Capping, covers, and grading,

- Diversionary structures,
- Excavation and disposal of solid mine waste,
- Re-use and reprocessing technologies.

These technologies will each be incorporated into closure planning to varying degrees, but will not be considered in the current evaluation of candidate passive treatment technologies. Together the incorporation of these "best practices" result in MIW that will be amenable to passive treatment.

The reasons for this re-evaluation of passive treatment technologies include:

- The technologies were last formally evaluated in connection with the Minto Phase IV mine closure plan. This proposed Phase V/VI Mine Plan has a revised configuration and some changed conditions;
- The YESAB evaluation of the Phase V/VI mine and closure plans contains guidance on how passive treatment should be evaluated and implemented (see Section 3);
- Minto Exploration Ltd (Minto) has committed to work collaboratively with Selkirk First Nation (SFN) on Closure Planning and water quality issues and to seek opportunities for involvement of SFN in both closure implementation and long-term care and maintenance; and
- The process can be used to foster a collaborative approach to optimization of the mine closure configuration. This is consistent with the recognition that Minto mine is in SFN traditional territory and will ultimately be returned to the SFN upon cessation of the mining and full, successful implementation of the RCP.

Specifically, the information presented in this memo is intended to be consistent with "Step 1 – Confirm Technologies" as outlined in an associated memo "Framework for Development of Minto Closure Water Quality Objectives" (August 5, 2015), and incorporates subsequent feedback from SFN related to the proposed process.

As described in the WQO Framework memo, scoping level work to date suggests that there is a good potential that some aqueous constituent load reductions from the site can be expected with passive treatment, but it is possible that these measures will not result in achieving a non-degradation target in lower Minto Creek. Minto is committed, in consultation with SFN, to exerting the best effort possible in refining a reasonable and practical passive treatment plan for Minto. In turn, objectives that are reflective of this plan and protect aquatic resources in lower Minto Creek will also be developed in consultation with SFN.

Minto recognizes the importance of evaluating the achievability of meeting non-degradation targets as outlined in the recent Water Licence Amendment. Minto, with the support of Alexco Environmental Group (AEG), has completed some additional analysis regarding the achievement of non-degradation targets. This analysis (results provided separately) suggest that although some aqueous constituents may reasonably be expected to meet a non-degradation concentration post-closure, many key constituents will not. As a result, the steps and evaluation in this framework remain focused on the key parameters considered previously, and also includes the parameters identified for water quality objectives in the revised Water Use Licence (operational water quality objectives, see section 3.1 below).

Given this introductory information, Section 2 of this memorandum presents background of how the passive treatment technologies proposed in previous versions of the Minto RCP came to be selected, and outlines the ongoing work being undertaken to provide site-specific proof of concept and develop the design basis. Given the changes in the mine plan and closure configuration for the proposed Phase V/VI mine expansion and our improved understanding of contaminant removal mechanisms that have been gained through the ongoing reclamation research program, this document also presents a framework for evaluating the relevance and applicability of a full range of potential passive treatment technologies.

Section 3 of this document then presents the updated ranking and evaluation of the various potential passive treatment technologies and identifies the passive (and semi-passive) treatment technologies that are proposed for consideration in closure planning at the Minto Mine Site. Section 4 identifies how the findings are proposed to be utilized, as next steps.

2 PREVIOUS WORK ON PASSIVE TREATMENT TECHNOLOGY SELECTION

A process for the identification and evaluation of the full range of passive treatment technologies that could potentially be incorporated into closure planning at the Minto Mine site was formally initiated in late 2012, in connection with the preparation of RCP v4.0. The work was initially presented to SFN in the associated preliminary FMEA workshop (January 2013). Prior to this formal process being undertaken, identification of potential passive treatment technologies was incorporated into closure planning at a conceptual level.

At that time, there was already substantial information available with respect to both observed and predicted water quality across the mine site and the anticipated long-term geochemical evolution of MIW. The need for a more formalized approach to identifying potential passive treatment technologies was recognized and the process was initiated in connection with early planning for the RCP v4.0. The framework utilized in the 2012/13 evaluations is outlined below in section 2.1, and is followed by brief summaries of the 2012/13 findings in sections 2.2 and 2.3 respectively.

2.1 FRAMEWORK FOR TECHNOLOGY EVALUATION

The options evaluation for closure reclamation technologies for the RCP v4.0 was based, in large measure, on the Interstate Technology and Regulatory Council (ITRC) mine waste treatment technology selection process, which is described in detail on the ITRC website: http://www.itrcweb.org/miningwaste-guidance/technology_overviews.htm

The ITRC is a public-private coalition and the technical material is developed by teams composed of environmental professionals, including state and federal environmental regulators, federal agency representatives, industry experts, community stakeholders, and academia. This site is continually updated and serves as an aggregator of best practice and innovative emerging treatment technologies and provides relevant case studies to support the design process. The ITRC mine waste technology selection process is recognized as a world class resource for on the topics of innovative mine reclamation and treatment technologies. As such, it was selected to use as the primary tool for identifying and evaluating potential treatment technologies.

The ITRC web-based resources on mine waste treatment technology selection process are designed to help regulators, consultants, industry, and stakeholders in selecting an applicable technology, or suite of technologies, which can be used to remediate mining sites. Through a series of questions, decision trees guide users to a set of treatment technologies that may be applicable to a particular site situation. Each technology is described, along with a summary of the applicability, advantages, limitations, performance,

stakeholder and regulatory considerations, and lessons learned. Each technology overview links to case studies where the technology has been implemented.

The decision tree for identifying potential treatment options for mitigating mining impacted groundwater and surface water is presented in Appendix A.

2.2 2012 PASSIVE TREATMENT OPTIONS ASSESSMENT

The 2012 Options Assessment considered the ability of various treatment technologies to treat observed and predicted water quality emanating from seeps at the South West Dump (SWD) and monitoring point W37, immediately downgradient of the dry stacked tailings storage area (DSTSF). The technology evaluation focused on treatment technologies that could address mining impacted water (MIW) as either surface water or groundwater flow.

The 2012 Options Assessment ranked the technologies in a semi-quantitative matrix, using a variety of metrics including:

- Ability to address main contaminants of potential concern (CoPCs; e.g., Cu, Cd, Se),
- Ability to operate in the climate and seasonality of water at Minto,
- Requirement for addition of chemical reagents,
- Ability to operate without constant power source,
- Successful track record in similar conditions,
- Opportunities to engage stakeholders,
- Long-term operational requirements,
- Long-term maintenance requirements, and
- Aesthetics.

The 4 short-listed technologies identified in the 2012 evaluation were: 1) constructed wetland treatment systems (CWTS), 2) biochemical reactors (BCR), 3) permeable reactive barriers (PRB), and 4) in-pit chemical precipitation treatment. Some of these technologies were further evaluated for potential applicability at Minto through the reclamation research program (refer to Section 2.4) to further refine the identification of the most promising passive treatment technologies for incorporation into long-term closure planning.

2.3 JANUARY 2013 PRELIMINARY FAILURE MODE AND EFFECTS ANALYSIS (FMEA)

The ITRC mine waste treatment technology selection process (web based decision tree) was presented at the January 2013 Preliminary Failure Mode Effects Analysis (FMEA) workshop along with the rationale by which the most promising passive treatment technologies were identified (i.e., CWTSs, BCRs, PRBs, and pit lake treatment). The potential application of these passive treatment technologies was then brainstormed in the context of evaluating a range of closure scenarios:

Scenario 1 – a source control focus in which emphasis would be placed on installing high quality, low permeability covers and subaqueous disposal of PAG waste rock to limit contaminant loading at the source.

Scenario 2 – referred to as a treatment focus (active and passive), considering minimal covers and source term control but heavy emphasis on treating all MIW before it leaves the site.

Scenario 3 – a hybrid closure scenario that relied on moderate covers (using readily available cover materials and industry practices for grading and revegetation) and incorporation of passive treatment technologies (with active treatment as contingency) to further improve water quality and decrease contaminant loads leaving the site.

Scenario 3 was ultimately endorsed and closure concepts including revegetated soil covers and passive treatment technologies were further developed in the Phase IV RCP.

It should be noted that subsequent to the 2013 FMEA, the site-wide geochemical modeling inputs were updated with new information and resulted in greatly improved estimated closure water quality relative to the working assumptions at the time of the 2013 FMEA.

2.4 PASSIVE TREATMENT RECLAMATION RESEARCH PLAN (RRP)

The results of the initial passive treatment technology evaluation, and subsequent feedback from the January 2013 FMEA workshop, were further considered by Minto and subject matter experts from AEG and Contango to develop a robust reclamation research plan for passive water treatment (RRP). The goal of the ongoing RRP is to enable, within the site-specific context of the Minto project, the determination of 1) proof of concept of each technology, 2) the anticipated treatment potential and 3) the design parameters required for more detailed design and costing of the proposed passive treatment technologies at closure.

The RRP for passive water treatment was initiated in 2013 and was designed to focus on CWTS and BCRs. The passive treatment RRP was designed to provide additional insight into innovative ways in which CWTS and/or BCRs could be applied at site, and potential to be combined as a treatment train to provide

maximum flexibility for a wide range of potential closure scenarios. Studies to evaluate the potential for batch treatment of pit-lakes (i.e., limnocorral trials) were already underway as that study had been initiated in the fall of 2012 (discussed below). Although permeable reactive barriers were initially identified as a potentially promising technology, they were not specifically evaluated through the RRP due to the anticipated challenging subsurface conditions (discontinuous permafrost) and limited areas of potential applicability. However, it was recognized that the proposed research into BCRs could provide relevant information to the potential applications and benefits of PRBs in a closure context.

The passive water treatment RRP included dedicated site visits with associated field and laboratory testing programs to provide preliminary characterization of potential treatment areas (Contango 2014a). Additionally, locally available borrow sources were sought out for substrates and vegetation that could be incorporated in passive treatment design. The initial field program identified areas where natural attenuation and treatment of MIW are already naturally occurring on-site and provided additional insight into potential treatment mechanisms. The approach being used to design the passive water treatment systems involves a phased approach, starting with off-site pilot-scale testing and optimization (at Contango's facilities in Saskatoon). The pilot-scale trials (medium-size trials in large barrels) included different combinations of soils/substrate and vegetation to allow for selection and optimization of the best design for the Minto site (Contango 2014b). The systems were designed to provide specific insights into the biogeochemical process-driven steps. The pilot-scale tests were fully instrumented, monitored regularly, and subjected to a variety of potential site conditions and water quality scenarios. The pilot-scale tests were run in a controlled facility, allowing for accurate evaluation of how different aspects affect the performance of the systems (e.g., nitrate, water depth) without the influence of other external and confounding factors. At the end of pilot-scale CWTS testing, additional organic material (straw and hay) were added to the wetland cells converting them to a hybrid CWTS/bioreactor. This provided a preliminary proof of concept for the feasibility of a semi-passive hybrid CWTS/bioreactor as a contingency option. The results of the pilot-scale testing were used to further optimize the design, and enabled refinement of sizing and set design, performance, and timeline expectations for the scale-up to a larger demonstration-scale treatment wetland to be constructed at Minto.

In the fall of 2014, an on-site demonstration-scale CWTS was constructed on the Minto Site to evaluate the effectiveness for treating actual MIW. The demonstration-scale system is constructed as four cells (two systems in parallel, each made of two cells in a series). The demonstration-scale CWTS has been progressing through commissioning as expected based on the pilot-scale testing, and is beginning to treat water (Contango, 2015; Capstone Mining, 2015).

Concurrently, dedicated research into the potential effectiveness of bioreactors has been undertaken at Yukon College both in lab facilities and at the Minto site. Lab-based testing utilized synthetic effluent to examine metals removal in sulphate-reducing conditions; these results are documented in Janin and

Harrington (2014). On site testing provided water from W37 to a series of reactors which utilized wood chips, ethanol, or wood biochar as substrates with gravel. Removal of selenium, copper, and nitrogen compounds was evaluated over a 2+ month operating window, and showed effective removal for both selenium and copper, and no detrimental effects to nitrogen compounds.

Limnocorral studies were undertaken prior to the development of the RRP and so it was not necessary to include additional pit-lake treatment studies as part of the RRP. In brief, to evaluate the potential for batch treating open-pit water, limnocorral studies were initiated in the Main Pit at Minto in October 2012 and ran through to the spring of 2013. The trials evaluated the feasibility and effectiveness of batch treating pit-lake water by addition of select carbon sources (sugars, alcohols and wood chips) to create reducing conditions (i.e., negative oxidation-reduction potential) that would in turn facilitate the precipitation of metals as sulphides and selenium as elemental selenium. Initial limnocorral tests provided promising results for metals treatment. If selected as a technology to pursue for implementation, additional research would be required to further develop the site-specific approach for batch treatment of the pit-lake water at Minto.

The passive water treatment RRP was initially designed to address the anticipated closure water quality conditions of the Phase IV mine plan. The proposed Phase V/VI Mine Plan contemplates changes to both the closure water conveyance network and to potential sources of contaminant loading. Consequently, the range of potential passive treatment technologies that might be applicable to the updated mine configuration will be evaluated anew.

3 UPDATED EVALUATION OF POTENTIAL PASSIVE TREATMENT TECHNOLOGIES

The Decision Document issued following the YESAA Screening of the Minto Mine Phase V/VI Expansion project proposal provided the following guidance for the development of closure WQOs:

> *#33. …Non-degradation (compared to historical background quality) of Minto Creek water quality shall provide the basis for the development of water quality objectives for the closure period. However, if non-degradation cannot be achieved using reasonable and practical passive treatment mitigations, then the closure objective shall be guided by what can be practically achieved (as long as the objectives are below the effects levels for aquatic resources with sufficient contingency). Determination of "reasonable" and "practical" mitigations must take into account the expected or actual site performance of a given mitigation, and the cost of the mitigation (both initial cost and long-term maintenance cost) compared to the expected contaminant reductions.*

This memo is designed to identify which particular technologies are considered as a 'reasonable and practical' basis of a long-term site-specific passive water treatment implementation strategy in the context of the Minto site. What is considered 'reasonable and practical' will depend on site-specific considerations and expectations of long-term operational and maintenance activities. For the purposes of evaluating and refining technology selections for passive water treatment and the Minto site, 'reasonable and practical' has been defined as having the following criteria:

- Operationally passive:
    - Self-sustaining in the long-term (e.g. minimal need for electricity or chemical addition)
    - Minimal ongoing operational oversight required
    - Does not require active decommissioning activities
- Minimizes long-term maintenance:
    - Able to naturalize, with only periodic and limited maintenance requirements
    - Most maintenance can be performed manually (i.e., does not require heavy equipment)
    - No routine addition of substrates or organics

In this context, it is recognized that even passive treatment technologies will require some operation and maintenance during commissioning. Moreover, several technologies under evaluation can span between passive to active, depending on their approach to implementation. For technologies that may be

implemented in different forms, only the most passive options are being considered as priority options in terms of being 'reasonable and practical' for the Minto site.

An updated reclamation and closure plan (RCP) was prepared for the Phase V/VI expansion. A Failure Modes and Effects Assessment (FMEA) workshop, attended by representatives from MEL, YG and SFN, was conducted in the summer and fall of 2014 to evaluate the latest iteration of the RCP (v5.1). The FMEA identified potential risks associated with closure water quality and also identified a need to address the topic of defining "reasonable and practical" passive treatment technologies to assist with the development of closure WQOs.

It was recognized at that time that a complete reassessment of the full range of potential passive treatment technologies was warranted, to ensure that any changing site conditions and/or emergent technologies were factored into the potential passive water treatment technology selection. A re-visiting of relevant site conditions was undertaken (Section 3.1) and the ITRC Process was then re-applied (Sections 3.2 – 3.5).

3.1 SUMMARY OF SITE CONDITIONS RELEVANT TO TREATMENT TECHNOLOGY EVALUATION

The following section briefly summarizes the specific site conditions that are directly relevant to closure planning and passive water treatment planning at the Minto mine site:

- Remote: The site is relatively remote. Accessing the site requires crossing the Yukon River via barge / ice dam or flying in via plane or helicopter. Once the closure plan is fully implemented, the airstrip would be decommissioned and barge service discontinued. Remobilization of heavy equipment would require temporary remobilization of the barge.

- Climate: The site is located in the sub-arctic ecoregion and experiences extreme cold during the winter months and a pronounced spring freshet.

- Contaminant Sources: The primary sources of ongoing contaminant loading in closure include waste rock dumps, dry stacked tailings, historic ore and concentrate stockpile areas and pit walls. Of these potential contaminant sources, the Southwest Dump and Drystack Tailings (DSTSF) account for the majority of predicted contaminant loads. The associated seepage from these contaminant sources has been monitored for years and the observed contaminant loadings add a significant degree of confidence to water quality predications.

- Site Configuration: The site is essentially confined to a single watershed which progressively constricts to a single point of discharge downgradient of the DSTSF (current spillway through the Water Storage Pond Dam). The majority of mining impacted water upgradient of the DSTSF is

routed through completed open-pits (which provide flood attenuation and additional residence time).

- Natural Attenuation: Significant reduction in contaminant loading is observed in the "wetland" area adjacent to sampling point W15. The observed contaminant load reduction in this flooded, natural area provides strong evidence that naturalized wetlands have the potential to provide ongoing passive treatment to mining impacted waters on site.

- Timing of Contaminant Flux: Evaluation of flow and concentration profiles from the mine has shown that peak flows typically occur in April and May, and that the metals loading distribution follows flow.

- Contaminants of Potential Concern: Aqueous metal and metalloid concentrations in site seepage and runoff are the primary focus of planned load reduction through passive treatment, as N-species (NH4, NO3 and NO2) concentrations are not anticipated to persist in post-closure conditions as they are a product of active operational blasting activities, and decay relatively quickly after these activities cease. The remaining COPCs identified in the current water use licence as parameters for which there is an operational water quality objective are:

    - Aluminum
    - Arsenic
    - Cadmium
    - Chromium
    - Copper
    - Iron
    - Lead
    - Molybdenum
    - Nickel
    - Silver
    - Selenium
    - Zinc

  These are the contaminants of focus for the ongoing evaluation of passive treatment options at Minto.

3.2 REVIEW OF THE ITRC MINE WASTE TREATMENT TECHNOLOGY DECISION FRAMEWORK

The ITRC mine waste treatment technology decision framework currently includes a total of 22 potential technologies (prevention and treatment) that could be applied to mitigate and/or treat mining impacted surface water and ground water. The list of technologies is reviewed and updated frequently by the ITRC

and is comprehensive, spanning operational life and mine closure, and includes both passive and active technologies.

The list of potential treatment technologies includes several active treatment technologies that do not meet the initial screening criteria (Section 2.1) of requiring no consistent power supply and a need to have low or no operation and long-term maintenance requirements. The active treatment technologies eliminated include: electrocoagulation, electrokinetics, ion exchange, and pressure driven membrane separation (e.g., reverse osmosis).

3.3 ADDITIONAL RESOURCES CONSIDERED

In addition to the ITRC decision framework referenced above, the following resources were also reviewed to ensure that all relevant passive treatment technologies have been considered:

1) Mine Environment Neutral Drainage guidance documents: http://mend-nedem.org/guidance-documents/
2) GARD guide: http://www.gardguide.com/
3) Alberta Environment published a review report titled “Evaluation of Treatment Options to Reduce Water-Borne Selenium at Coal Mines in West-Central Alberta”.
4) CH2M Hill prepared a review report for North American Metals Council titled “Review of Available Technologies for the Removal of Selenium from Water”
5) Golder prepared a review report for Teck titled “Literature Review of Treatment Technologies to Remove Selenium from Mining Influenced Water”.
6) MSE prepared a report for US EPA and Department of Energy titled “Selenium Treatment/Removal Alternatives Demonstration Project”.

3.4 LONG LIST OF POTENTIAL PASSIVE TREATMENT TECHNOLOGIES

The long list of potential MIW mitigation methods/technologies is presented in Table 1. This table addresses each technology at a high level, provides general comments about the applicability of each technology to Phase V/VI closure planning, the current status of evaluation of the technology or approach, and how that technology could potentially fit into Minto-specific closure planning.

As noted in Section 1, six (6) of the seventeen (17) long-listed methods/technologies are already incorporated in the existing designs of Minto Mine closure. In addition, six (6) are considered not relevant to site conditions at Minto (aeration treatment systems, anoxic limestone drains, chemical stabilization, in situ biological treatment, passivation technologies, and phytotechnologies). The rationale for these decisions is provided in the attached tables. This leaves five (5) passive treatment technologies that are considered relevant in the context of identifying "reasonable and practical passive treatment" as defined in the Decision Document. These are:

- Constructed wetland treatment systems;
- Biochemical reactors;
- Permeable reactive barriers;
- In situ treatment of open pits (chemical precipitation through sulphate reduction); and
- In situ treatment of open pits (enhanced biological treatment using algae to sequester metals).

These five (5) treatment technologies are considered further as "short listed" potential passive treatment technologies.

3.5 SHORT LIST OF POTENTIAL PASSIVE TREATMENT TECHNOLOGIES

The short list of potential passive treatment technologies is outlined in Table 2. This list has been maintained intentionally broad in recognition that many of these technologies are "variations on a theme" and can be tailored and/or subject to further optimization in an integrated closure scenario.

Of the five (5) technologies that made the short-list, only one (1) has been identified as a primary passive treatment technology in the Minto Mine closure context – CWTSs. This is primarily based on the level of site-specific research that has been conducted at the Minto site, and the technologies' 'readiness' for advanced design- level integration into the current closure planning.

Four (4) additional technologies are presented as candidate contingency passive treatment technologies – BCRs, PRBs, and two variations of in-situ batch treatment in open pits. Batch treatment of open-pits is subdivided into sulphate reducing batch treatment (which removes metals by precipitation as sulphides) and enhanced biological treatment (which removes metal by algae sequestration which then sinks to the base of the pit). These measures can be considered at a 'contingency' plan level of integration into current closure planning at the Minto site, but would require further site-specific experimentation to be able to refine performance expectations.

4 PROPOSED PATH FORWARD

This document should serve as a foundation for the further development and refinement of:

1. Closure Water Quality Objectives – This technical briefing document is a reference for the ongoing collaborative work underway in the Closure WQO development. The next step (Step 2 – Develop Configuration Options) of the framework is underway, and preparations are being made for Step 3 (Select Configuration and Implementation Strategy); and
2. Refinement of the RCP v6.0 Reclamation Research Plan – this document in combination with outcomes of the WQO Framework will advise the intent and content of the research programs that will continue at the Minto Site.

5 REFERENCES

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TABLE 1
FULL LIST - ITRC MINE WASTE MITIGATION TECHNOLOGIES (PREVENTION AND TREATMENT)
RE-EVALUATION IN CONNECTION WITH PHASE V/VI RCP

TECHNOLOGY	CATEGORY (TREATMENT OR PREVENTATION)	BASIC DESCRIPTION	POTENTIAL APPLICABILITY FOR MINTO CLOSURE	STATUS OF EVALUATION	CLOSURE ROLE
Administrative and Engineering Controls	Prevention. (Already Incorporated in Closure Planning)	Administrative controls (ACs) are nonengineered instruments intended to minimize the potential for human exposure to contamination by limiting land or resource use. Engineering controls (ECs). ECs are physical controls put into place to prevent human and ecological exposure to contamination. Several other technologies listed in this table (e.g., capping, covers and grading; diversionary structures; backfilling and subaqueous disposal) also fall within the broad EC category.	Incorporated into all aspects of closure planning.	Key concerns were flagged in connection with ongoing planning and consultation, and most recently in connection with the Closure FMEA. Incorporated into AMP.	Incorporated into AMP as well as monitoring and maintenance planning. Does not play a role in the evaluation of "reasonable and practical" passive treatment.
Aeration Treatment Systems	Treatment of anoxic or high BOD waters. (Not Relevant)	Aeration involves the mechanical introduction of oxygen into the MIW stream through a variety of techniques with the goal of oxidizing specific dissolved metals species into less soluble forms. Aeration can use gravity and/or mechanical devices to increase the concentration of dissolved oxygen in MIW, promoting oxidation of ammonia, and metals and metalloids such as iron, manganese and arsenic. Some other metals species may also bind to the iron or manganese.	The elements treated by aeration are not relevant to the MIW at Minto. Generally not relevant to Minto unless BCRs or PRBs create a high BOD in outflow water, aeration could be incorporated (gravity flow) to reoxygenate water for receiving environment. Aeration could potentially play a role in oxygenating low oxygen waters to support aquatic ecosystem health.	The role of aeration will be considered in the context of the water chemistry and constituents needing removal from the water, in connection with water conveyance along primary ditches and adjacent to passive treatment zones.	Will be evaluated if needed downstream of BCRs or PRBs to reoxygenate waters. Design principles are well understood. No requirement for specific field trials.
Anoxic Limestone Drains	Treatment of acidic water. Semi-passive. Finite life-span. (Not relevant based on current/predicted water quality)	Anoxic limestone drains (ALDs) are passive treatment systems that can be used to treat the acidity of mine-influenced water (MIW) under specific geochemical conditions. ALDs consist of a buried bed of limestone ($CaCO_3$) engineered to intercept anoxic, acidic MIW and add alkalinity through dissolution of the limestone.	Not relevant to treatment of Minto MIW as the seepage is circumneutral. The technology is well understood by the closure design team and could be incorporated if acidic condiitons develop.	Not appropriate as a key technology for Minto (based on our understanding of the site wide geochemistry), but this technology is well understood by the design team and will be incorporated in the design of passive treatment systems if conditions warrant.	Not currently planned.
Backfilling and Subaqueous Disposal	Prevention. (Already incorporated in closure planning)	In its most basic form subaqueous disposal involves removal of surface material and placing it underground and/or under water, thus eliminating direct contact exposures. It is typically applied to sulfide-containing solid mine wastes to reduce oxidation of the wastes, thus limiting acid generation and/or neutral metals release. It has also been used to dispose of non-acid-generating solid mine wastes through backfilling.	This is being incorporate in the closure planning with tailings and ARD/ML waste rock being deposited in completed open pits which will flood and submerge the mine waste at closure	Saturation of PAG waste rock has always been a part of closure planning at Minto. Minto's Water Use Licence also contains specific language that requires that potentially reactive waste rock associated with the Phase V/VI mine expansion will remain saturated in perpetuity under closure conditions and requires further evaluation of maximizing the backfilling of tailings and mine waste in exhausted open-pits and mine workings.	Mine plan currently being revised to optimize backfilling and subaqueous disposal - consistent with WUL requirements. The implications of subaqueous disposal are already incorporated in WQ modeling and further optimization is not part of current discussion of additional "reasonable and practical" passive treatment technologies for developing closure WQOs.
Biochemical Reactors (BCRs)	Treatment. Semi-passive. Finite life-span. (Proposed in existing closure planning. Re-evaluated as a contingency technology due to finite life span)	Biochemical reactors (BCRs) treat mining-influenced water (MIW) by using microorganisms to transform contaminants and to increase alkalinity in the treated water. The most commonly used BCRs for treating MIW are operated anaerobically (no oxygen) and are also called “sulfate-reducing” bioreactors. The microbial process of sulfate reduction produces sulfide and bicarbonate within the reactor, allowing the target metals such as cadmium, copper, nickel, lead, and zinc in MIW to precipitate as metal sulfides at pH values above 5.0. Biological selenium reduction (treatment) also occurs under these conditions. The bicarbonate produced through sulphate reduction promotes an increase in pH and will promote the removal of some metals as carbonates such as $FeCO_3$ and $ZnCO_3$ under the appropriate conditions.	This technology is semi-passive, requiring periodic maintenance (e.g., injection of carbon source and/or periodic replenishment of porous media). As such, their applicability is considered to be limited to early in the closure period (i.e., PC1) after which time the BCR would no longer be maintained. The exhausted BCR would either be decommissioned or left in place in a geochemically stable condition. They could be installed as stand-alone treatment features (potentially to treat concentrated seepage at the toes of waste rock dumps or the DSTSF), but more likely in connection with a CWTS, where the CWTS would perform treatment once the BCR is no longer effective.	BCRs are being evaluated in connection with the ongoing RRP in coordination with Yukon College.	This research is being conducted in connection with the RRP. This is a semi-passive technology that could be employed early in the mine closure process with the intension that the system would be left and cease to be maintained beyond an initial operational period. This technology is considered as a contingency passive treatment technologiy for closure at Minto.

TABLE 1
FULL LIST - ITRC MINE WASTE MITIGATION TECHNOLOGIES (PREVENTION AND TREATMENT)
RE-EVALUATION IN CONNECTION WITH PHASE V/VI RCP

TECHNOLOGY	CATEGORY (TREATMENT OR PREVENTATION)	BASIC DESCRIPTION	POTENTIAL APPLICABILITY FOR MINTO CLOSURE	STATUS OF EVALUATION	CLOSURE ROLE
Capping, Covers and Grading	Prevention. (Already incorporated in closure planning)	Capping or covering of solid mining waste is an effective and proven treatment technology. Installation of a cap or cover on solid mining waste can reduce or eliminate erosion, fugitive dust emissions, and infiltration of water to prevent the migration of contaminants. Caps or covers eliminate direct exposure to solid mining waste by creating a physical barrier that prevents direct contact with the contaminants.	All mine waste will be capped with an isolating soil cover and revegetated to facilitate the naturalization of the site at closure and a return to self-sustaining conditions that are consistent with land use planning goals.	An evaluation of the full potential range of cover designs was completed by SRK in connection with closure planning and preparation of RCP v4.0. The evaluation concluded that the only cover material available in sufficient quantities for use as cover material is overburden from the stripping of the open-pits and mine infrastructure foundations.	All mine waste will be appropriately graded and covered with a minimum 0.5m of revegetated, isolating soil cover. The HGW portion of the SWD will be capped with a very low permeability BGM cover. The anticipated cover performance is already incorporated in the water quality modeling and is not being considered further in the current context of defining additional "reasonable and practical" passive treatment technologies.
Chemical Precipitation (e.g., In-Pit)	Treatment. Passive or Semi-passive if in-pit settling is designed rather than active filtration/removal of precipitates. (Addressed as a contigency treatment technology in current closure planning in the context of in-situ treatment of water in open pits)	Chemical precipitation is a conventional technology used to treat mining-influenced water (MIW), including acid mine drainage, neutral drainage, and pit lake water. Chemical precipitation processes involve the addition of chemical reagents, followed by the separation of the precipitated solids from the cleaned water. Typically, the separation occurs in a clarifier, although separation by filtration or with ceramic or other membranes is also possible. Chemical precipitation can also be used in pit lakes or other water bodies, in which case the precipitated solids can simply settle and remain in the bottom of the pool.	Potential for significant role as contingency for treating Area 2 Pit water quality. Potential for carbon loading to turn the pit water reducing (encouraging sulphate reduction) and precipitate metals (e.g. Cu, Cd) as sulphides, and reduce SeIV and SeVI to elemental selenium (Se0) which would all settle to the pit bottom. This treatment technology overlaps with In-Situ Treatment of Pit Lakes (below).	Limnocoral trials were completed over the fall/winter of 2013/14. Evaluation to date has focused on creating sulphate reducing conditions to precipitate metals as sulphides (with an emphasis on treatment of Se). Batch treatment of Open-Pits is incorporated in the RRP. Limnocoral trials have confirmed the presence of selenium reducing bacteria and provided proof of concept. Currently considered a contingency treatment technology. The potential application of enhanced biological treatment (algae precipitation) is relatively well understood based on the Grum Pit trials at Faro.	Preliminary limnocoral trials in main pit provided proof of concept. Additional study (limnocoral) will likely be undertaken as Area 2 Pit begins to fill towards the end of the mine life. It is anticipated that batch treatment of Area 2 Pit will remain a contingency plan for closure. No requirement for additional study at this time.
Chemical Stabilization, Phosphate and Biosolids Treatment	Prevention. (Not practical at Minto)	Chemical phosphate treatments have used a variety of phosphate species, but phosphoric acid has been demonstrated to be the most effective. Organic sources of phosphate such as biosolids or composted animal wastes have also been used to stabilize, reclaim, and revegetate barren mine and mill wastes.	Not considered at the Minto Site due to a lack of readily available, cost effective source materials. Other, more appropriate treatment options are readily available.	Not considered for application at Minto.	Not applicable. Dropped from further consideration.
Constructed Wetland Treatment Systems (CWTS)	Treatment. Passive or Semi-passive. Can be designed for long-term, perpetual treatment. (Proposed in existing closure planning)	Constructed treatment wetlands are man-made biologically active vegetated systems that are characterized by saturated soil conditions and at least periodic surface or near-surface water designed specifically to treat contaminants in surface water, groundwater, or waste streams. The wetlands can be designed to operate either aerobically, or anaerobically, depending on the water treatment requirements. The CWTS can be operated as part of a treatment train (e.g., after BCR or in pit treatment).	Several areas are currently being considered. The most significant area is immediately downgradient of the DSTSF and MVFE (treating W37 seepage) and the area around W15. The area around W15 is already functioning as a natural treatment wetland. Evaluations are ongoing for ways to further enhance treatment peformance. Open pit overflow (i.e., Pit 2) is also being considered for treatment.	One of the primary passive treatment technologies being evaluated for Minto. Described in detail in the Reclamation Research Plan. Detailed pilot-scale trials specific to Minto have already demonstrated proof of concept and anticipated removal rates for CoCs for a given surface area, as well as developped the preliminary design considerations. Minto is currently operating an on-site demonstration CWTS to optimize design and refine full-scale sizing requirements.	Will be incorporate into long-term treatment of MIW. Next step is to confirm specific areas for implementation on the mine plan (closure configuration) and confirm available footprint and anticipated outflow water quality based on removal rates (developed from demonstration scale wetland), inflow water quality, and flow rates.
Diversionary Structures	Prevention. (Included in existing closure planning)	Diversionary structures are designed to prevent clean water from becoming MIW by coming into contact with mining solid waste (net acid-producing materials) and/or to divert MIW to treatment or collection systems and away from sensitive environments. Diversionary structures can be used to reduce the volume of, or exposure to, MIW that may present risks to human or ecological receptors and also to prevent/reduce erosion.	Incorporated where possible to both protect mine closure infrastructure and minimize contact of clean water with mine waste. Examples include the Tailings Diversion Ditch and the various primary and secondary water conveyance ditches.	Diversionary structures are already designed for the currently anticipated Phase V/VI closure configuration.	Incorporated in closure planning as appropriate. Concepts are well understood and there is no requirement for additional research in connection with the development of WQOs.

TABLE 1
FULL LIST - ITRC MINE WASTE MITIGATION TECHNOLOGIES (PREVENTION AND TREATMENT)
RE-EVALUATION IN CONNECTION WITH PHASE V/VI RCP

TECHNOLOGY	CATEGORY (TREATMENT OR PREVENTATION)	BASIC DESCRIPTION	POTENTIAL APPLICABILITY FOR MINTO CLOSURE	STATUS OF EVALUATION	CLOSURE ROLE
Excavation and Disposal of Solid Mine Waste	Prevention. (Included in existing closure planning. Largely duplication of Backfilling/ Subaqueous disposal category)	Soil, sediment, or tailings can be removed so that the remaining contaminant concentrations meet a risk-based cleanup level or removed to a certain depth or areal extent so that the clean backfill placed on top of the remaining contamination creates a physical barrier preventing direct contact with the contaminants.	This is already incorporated into closure planning to the degree practicable in the form of excavating pockets of contamination (e.g., areas around concentrate piles or Pelley bone-yard) and dispolsal in engineered landfills or with other reactive mine wastes as appropriate. Application to larger volumes of contaminated mine wastes (e.g. waste rock) is generally covered under backfilling and subaqueous disposal of mine waste (described above) and is a specific requirement within the most recent WUL conditions.	Mine plan is currently being updated to ensure consistency with the WUL requirements. PAG rock that is excavated during the next mine phase will be stored in areas that will be saturated under closure conditions.	This is already incorporated in closure planning ("good practice").
In-situ Biological Treatment	Treatment. Semi-passive. Finite life-span. (Not relevant as defined herein, although partially overlaps with the category chemical precipitation)	In situ biological source treatment consists of isolating the source of mining-influenced water (MIW) through the establishment of an in situ biological layer on exposed metal sulfide surfaces. This is typically accomplished through the injection of inoculum (e.g., wastewater effluent) and substrate into the subsurface material. The in situ biological source treatment can achieve satisfactory results without the cost of excavation and material handling.	Not specifically considered for Minto – although this technology is closely related to BCRs and CWTS which are considered more appropriate technologies for closure application at Minto.	No specific research has been completed. Not specifically considered for closure applications at Minto.	No. Not considered for closure application at Minto.
In Situ treatment of Mine Pools and Pit Lakes	Treatment. Passive or Semi-passive. Can have long-lasting effects, depending on source water and pit characteristics. (Proposed as contingency measure in existing closure planning)	The technology consists of the injection or placement of substances, including (as appropriate) alkaline materials and organic carbon substrate, with nutrients directly into the mine pool or pit lake to neutralize the MIW and to produce anaerobic conditions to precipitate metals in place. Injection of a carbon source such as molasses or alcohol with nutrients and sometimes an alkaline source, such as lime, can create conditions favorable to the precipitation of dissolved metals in place.	Treatment of pit lakes is being evaluated at Minto and is a part of the ongoing RRP. The mine pools are not anticipated to require any treatment. A variation on a theme, which has not been previously discussed for Minto, is encouraging algal growth (through the addition of fertilizer/nutrients) which would scavenge metals and precipitate to the bottom of the pit (e.g., Similar to Grum Pit trials at Faro). Potential challenges with this include potential bioaccumulation of Se as it is taken up by algae; therefore this technology could likely only be considered in combination with initial sulphate reduction to first remove Se from the water column. Enhansed algal growth could be used to maintain treatment conditions semi-passively.	Limnocoral trials were completed over the fall/winter of 2013/14. Evaluation to date has focused on creating sulphate reducing conditions to precipitate metals as sulphides (with an emphasis on treatment of Se). Batch treatment of Open-Pits is incorporated in the RRP. Limnocoral trials have confirmed the presence of selenium reducing bacteria and provided proof of concept. Currently considered a contingency treatment technology. The potential application of enhanced biological treatment (algae precipitation) is relatively well understood based on the Grum Pit trials at Faro.	In-situ treatment of pit lakes (specifically the Area 2 Pit) is being considered as a contingency measure for closure .
Passivation Technologies	Prevention. (Considered not relevant / not practical at Minto)	Technology involves chemically producing coatings on reactive mine waste to limit metals mobility. Typical solutions include use of phosphates and silica. Treatment of acid-generating rock with permanganate and magnesium oxides at a high pH (>12) is a patented process. All passivated surfaces still have reactive rock below the surface, and oxidations will return once that passivation layer is removed. Thus, whether passivation is a viable option depends on time, other environmental conditions, and treatment efficiency requirement.	Not specifically considered in closure planning to date. May have application for treatment of pit walls in the event that ML from exposed surfaces proves problematic in the long-term. Geochemical modelling to dateprovides a high degree of confidence that metal leaching from exposed pit walls at closure is unlikely to be problematic. Passivation technologies typically produce inconsistent and relatively short duration reduction in contaminant loading.	Additional primary research would be required in advance of any large scale trials.	Not explicitely considered in the current RCP.
Permeable Reactive Barrier (PRB)	Treatment. Semi-passive. Finite life-span. (Proposed in existing closure planning. Re-evaluated as contingency treatment technology due to finite life-span and limited applicability in a Minto specific closure context)	In the broadest sense, a permeable reactive barrier (PRB) is a continuous, in situ permeable treatment zone designed to intercept and remediate a contaminant plume. The treatment zone may be created directly using reactive materials such as iron or indirectly using materials designed to stimulate secondary processes, such as by adding carbon substrate and nutrients to enhance microbial activity.	PRBs could potentially be incorporated in very specific applications, as part of a passive treatment train, in areas where space is limited and the potential exists to passively intercept mining impacted groundwater (e.g. immediately downgradient of the DSTSF.	The RRP research into bioreactors and CWTSs, while not explicity directed towards the design of PRBs, provides significant insight into the design and operation of PRBs (i.e., hydraulic, geochemical and microbial characteristics) in a Minto closure context.	This is considered a variation on a theme similar to BCR. Recognized that PRB would likely be used as part of a treatment train rather than as a stand-alone technology.

TABLE 1
FULL LIST - ITRC MINE WASTE MITIGATION TECHNOLOGIES (PREVENTION AND TREATMENT)
RE-EVALUATION IN CONNECTION WITH PHASE V/VI RCP

TECHNOLOGY	CATEGORY (TREATMENT OR PREVENTATION)	BASIC DESCRIPTION	POTENTIAL APPLICABILITY FOR MINTO CLOSURE	STATUS OF EVALUATION	CLOSURE ROLE
Phytotechnologies	Prevention. (Not relevant in WQO context)	Phytotechnologies use plants to remediate various media impacted with different types of contaminants. Phytotechnologies can be applied to address certain issues associated with mining solid wastes and mining-impacted waters. There are six basic phytoremediation mechanisms that can be used to clean up mining-contaminated sites: phytosequestration, rhizodegradation, phytohydraulics, phytoextraction, phytodegradation, and phytovolatilization	All mining waste will be covered with a minimum 0.5m of clear overburden and revegetated. Consideration of specific phytotechnologies, such as phytosequestration and phytohydraulics may be warranted in specific, small scale applications, such as seepage zones, but generally in the context of protection of wildlife and the prevention of metal uptake by plants that could be detrimental to wildlife that consumes those plants (i.e., not to remediate metals in soil).	Not specifically evaluated as a primary reclamation technology for Minto but understanding of the topic (phytosequestration, rhizodegradation, phytohydraulics, phytoextraction, phytodegradation and phytovolatilization) is incorporated in a general sense within the revegetation evaluations.	Consideration of specific phytotechnologies, such as phytosequestration and phytohydraulics may be warranted in specific, small scale applications such as seepage/wet zones – but generally in the context of protection of wildlife and the prevention of metal uptake by plants that could be detrimental to wildlife that consumes those plants.
Re-Use and Reprocess (R^2) Technologies	Prevention. (Incorporated in existing closure planning)	Re-use consists of using problematic mine waste either directly or following reprocessing or other treatment as a beneficial product that is environmentally safe in its re-used form. Reprocessing consists of subjecting mine waste to physical or chemical processes designed to extract minerals or other waste components for beneficial use, rendering the waste material suitable for other beneficial use or environmentally safe disposal on the mine site.	Marginally economic waste rock will be processed as economically feasible. The site is remote site has limited potential for re-use (e.g., use of clean waste rock as rip-rap in streams has proven prohibitively challenging). The WUL specifically prohibits mine waste from being transported off-site.	Optimal reprocessing of marginal waste rock is incorporated in the MWROMP. No further evaluation planned.	Further optimization is not considered. Topic included for completeness.

TABLE 2
SHORT LIST - ITRC PASSIVE TREATMENT TECHNOLOGIES
RE-EVALUATION IN CONNECTION WITH PHASE V/VI RCP

TREATMENT TECHNOLOGY	Status of Evaluation	Steps to Advance	Site Considerations; Reasonable/Practical	Potential Areas of Application	Opportunities for SFN Participation	Key Documents
Primary Passive Treatment Technologies [1]						
Constructed Wetland Treatment System (CWTS)	Highly advanced as documented in the RRP. A field scale CWTS is currently in operation at the Mine Site with the option of converting to a hybrid CWTS/BCR as a contingency. The flooded area around W!5 has also been evaluated in the context of natural attenuation of contaminant loading that is already occuring (in a non-engineered setting) and the potential for enhanced natural attenuation through further optimization and plant species and improved routing of flows.	This is the primary passive treatment technologies being evaluated for Minto. Described in detail in the Reclamation Research Plan. Potential to be converted to a semi-passive BCR if needed as a contingency. Potential sites have been identified and predicted water quality will be refined as data continues to be produced form the on-site demosntration CWTS. Steps 2 and 3 of the "framework for developing Minto closure WQOs" will involve identifying potential sites for incorporating CWTSs (alone or as part of treatment train) under closure conditions.	This is considered both reasonable and practical at the Minto Site and is summarized in Contango (2014a, 2014b and 2015)	Anywhere MIW surface water can be routed through a relatively flat area of sufficient size to accomodate a treatment cell. Whenever possible, should be installed as multiple systems at water sources to reduce loads at the source. A primary area under consideration is immediately downgradient of DSTSF and MVFE. Additionally, the wetland in the W15 area is already naturally functioning in many respects to treat water and could be further enhanced at closure.	Sourcing wetland vegetation. Producing elements of substrate (e.g. wood chips, straw). Construction of the CWTS and monitoring and maintenance during commissioning.	RRP, CWTS site assessmet, CWTS pilot-scale testing, CWTS field-scale construction documents.
Contingency Passive Treatment Technologies [2]						
Biochemical Reactor (BCR)	Somewhat advanced as documented in the RRP. A small-scale BCR is currently in place on site under the design of Yukon College.	Further on-site investigations of seepage sources (quality, variability, flow rates, seasonality) should be undertaken to inform further in-situ research, including demonstration scale reactors. This research would further inform the feasibility of the use of BCRs in the closure planning at Minto.	This technology is semi-passive, requiring periodic maintenance (e.g., injection of carbon source and/or periodic replenishment of porous media). As such, their applicability is considered to be limited to early in the closure period (i.e., PC1) after which time it would no longer be maintained and would be isolated and remain buried in geochemically stable condition. BCRs could be installed as stand-alone treatment features, but more likely in connection with a CWTS, where the CWTS would perform treatment once the BCR is no longer effective. Because BCR are only semi-passive and shorter life span than CWTS, they are most practical to apply as a secondary option where insufficient footprint is available for a CWTS, or concentrations are higher than could be treated by a CWTS. This will need to be evaluated against findings for CWTS feasibility	These are being evaluated in connection with the ongoing RRP and evaluation of CWTS (i.e., to determine which sites would have sufficient area for effective CWTS treatment, and which would need to consider a bioreactor for treatment in a smaller footprint). Potential exists to turn area below the MVFE into a BCR by injecting carbon source to generate sulphate reducing conditions. Still at conceptual level - considered contingency application. If this is done, the CWTS design at that location may need to be revisited as water chemistry will have different treatment requirements (e.g., BOD)	Producing elements of porous media/substrate (e.g. wood chips, granular…). Participation in the construction, monitoring and maintenance during commissioning, and ongoing operation, monitoring, and maintenance.	RRP
Permeable Reactive Barrier (PRB)	This is considered a variation on a theme similar to BCR. It is being evaluated in connection with the RRP. Recognized that PRB would likely be used as part of a treatment train rather than as a stand-alone technology.	Focused review of subsurface information would be required prior to any detailed planning associated with PRBs.	Practical only where groundwater plume is identified and constrained into a groundwater source area that can be channelled.	Best suited where space is limited and groundwater must be passively intercepted/treated - such as immediately downgradient of DSTSF and MVFE.	Sourcing organics for permeable media. Monitoring of operational performance.	RCP v4.0. RRP
In-Situ Treatment of Open Pit (Chemical Precipitation through sulphate reduction)	Limnocoral evaluation has provided proof of concept - suitable for establishing anticipated removal rates of CoCs and required dosing with carbon sources.	Initial limnocoral trials in 2013/2014 have provided proof of concept. Additional trials may be completed as the Area 2 Pit begins to fill during the later stages of Phase V/VI operations.	Proof of concept demonstrated in Main Pit limnocorral trials. Would require mobilization of reagents and equipment for each implementations. Potential application in both Main Pit and Area 2 Pits.	Batch treatment of Open-Pits is incorporated in the RRP. Limnocoral trials have confirmed the presence of selenium reducing bacteria and provided proof of concept. Currently considered a contingency (semi)passive treatment technology.	Monitoring of pit water quality for trends in water quality that would warrant the implementation of batch treatment.	Limnocoral trials
In-Situ Treatment of Open-Pit (Enhanced biological Treatment using algae to sequester metals)	Has not been formally evaluated at this time. Risks identified with potential amplification of Se toxicity as algae could potentially make Se more bioavailable.	Internal discussion. Review the available data from the Grum Pit treatment trials at Faro (program overseen by Lorax). Research implications for Se removal.	Contimant load reduction has been demonstrated at similar sites (e.g., Faro Mine, Grum Pit).	Area 2 Pit. Contingency treatment technology. Would likely only be implemented in combination with batch treatment to drive sulphate reduction to remove Se.	Monitoring of pit water quality for trends in water quality that would warrant the implementation of batch treatment. Application of fertilizer/nutrients to promote algae growth.	Grum treatment evaluations from Faro (additional literature review warranted)

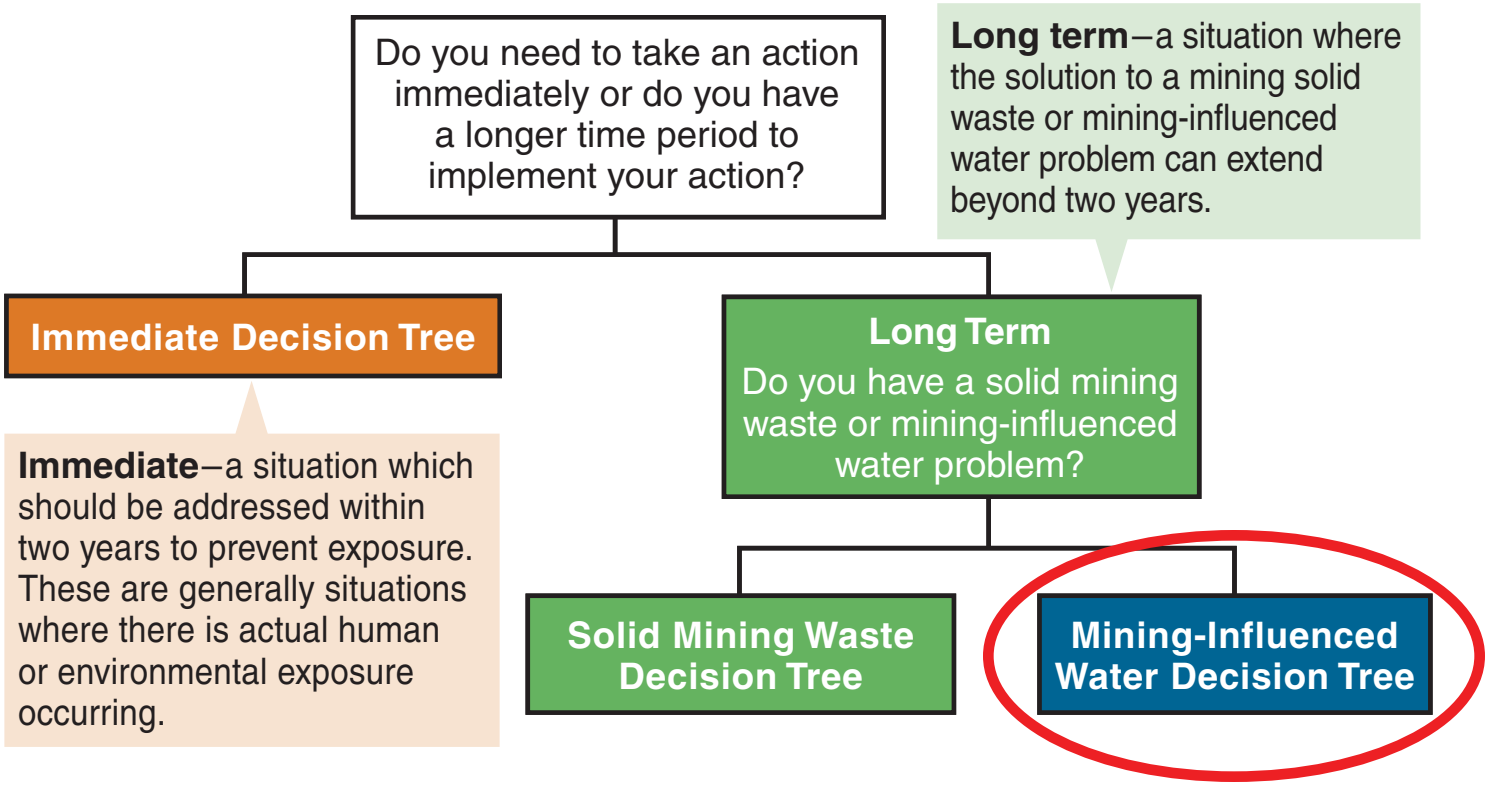
Notes:
1 This technology is considered the most appropriate for evaluation under "Steps 2 and 3" of the Framework for Developing Minto Closure WQO.
2 These technologies are considered "Contingency" passive treatment technologies.

Decision Tree

The Decision Tree has been designed to guide users to a set of treatment technologies that may be useful in managing a particular mine waste site. The user is presented with a series of questions. By answering the questions, the user is directed towards appropriate treatment technologies. Because of the size and complexity of most mine waste sites, there are generally a variety of environmental problems to be addressed. Those problems may include contaminated groundwater, contaminated residential yards, large areas of mine waste, or contaminated surface waters. The user should go through the decision tree separately for each issue to be addressed.

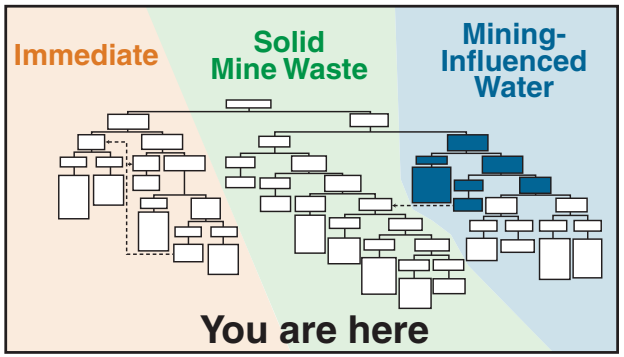
At the end of each string of questions in the decision tree is a list of treatment technologies. Clicking on that list will take users to Technology Overviews where they will learn more about the applicability of specific technologies and supporting case studies.

Mining Waste Team Decision Tree—Initial Questions



Mining-Influenced Water Decision Tree

Part One



Mining-Influenced Water

Do you need to control water quality at the human receptor or at the source?

Receptor

Administrative/ Engineering Controls

Pressure-Driven Membrane Separation

Ion Exchange

Source

Can you eliminate the mining-influenced water by addressing the solid mining waste source?

Yes

See Solid Mining Waste Decision Tree

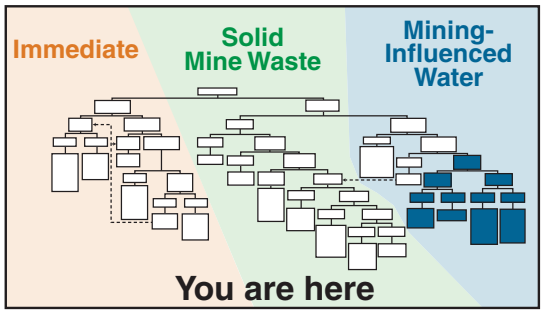
No

See Solid Mining Waste Decision Tree

Part Two

Mining-Influenced Water Decision Tree

Part Two



No

Do you need to control water quality in groundwater or surface water?

Groundwater

Do you want to pump water and treat it at the surface?

No

- ☑ Permeable Reactive Barriers
- ☑ In Situ Treatment
- Electrokinetics
- ☑ In Situ Biological Treatment

Yes

See Surface Water

Surface Water

Do you need a treatment technology that is more passive or can you use a more active technology?

Passive

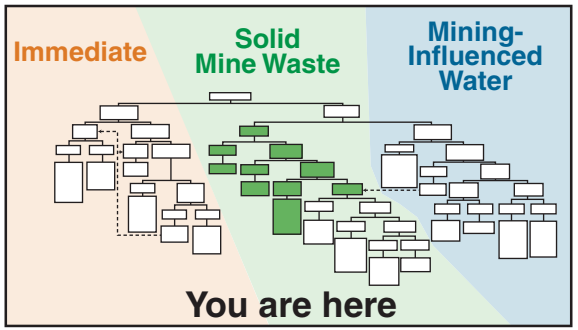
- ☑ Biochemical Reactors
- ☒ Microbial Mats
- ☑ Constructed Treatment Wetland
- ☒ Anoxic Limestone Drains
- ☑ Aeration
- ☑ In Situ Biological Treatment

Active

- ☑ Chemical Precipitation
- Ion Exchange
- Pressure-Driven Membrane Separation
- Aeration
- Electrocoagulation

Solid Mining Waste Decision Tree

Part One



Solid Mining Waste

Do you have saturated sediments affected by mine waste?

Yes

ITRC Contaminated Sediments Public Page

No

Do you need to control exposure to mining wastes which have been transported indoors?

Yes

Administrative/ Engineering Controls

No

Do you need to control exposure in a residential yard?

Yes

Administrative/Engineering Controls

Excavation and Disposal

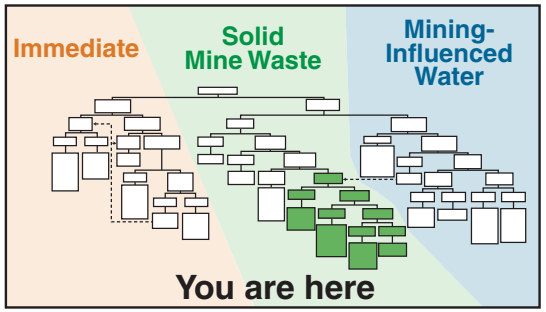
Capping, Covers and Grading

Chemical Stabilization

No

Solid Mining Waste Decision Tree

Part Two



From Mining-Influenced Water Decision Tree

No
Is it feasible to remove the mine waste?

Yes

- ☑ Excavation and Disposal
- ☑ Subaqueous Disposal
- ☑ Re-Use and Reprocess

No
Can you control exposure by treating the mining waste?

Yes

- ☑ Passivation
- Chemical Stabilization
- ☑ In Situ Biological Treatment
- Electrokinetics
- Phytotechnologies

No
Can you control exposure with physical barriers?

Yes

- ☑ Capping, Covers and Grading
- ☑ Diversionary Structures
- ☑ Phytotechnologies

No

- ☑ Administrative/ Engineering Controls

	Closure Scenario 1: Source Control Focus	Consequence		Likelihood	Risk Rating	NOTES
		Type	Severity	Probability	Descriptive	
1	**Source Terms**					
A	Source water quality (source term) worse than expected and causes unacceptable water quality conditions **downstream of site**	Env. Imp.	Critical	Possible	High	assuming no AMP in place
B	Source water quality (source term) worse than expected and causes unacceptable water quality conditions **downstream of site**	Env. Imp.	Moderate	Possible	Moderately High	assumed AMP in place so Severity moderate, critical to understand chemistry as fully as possible - reflected in likelihood designation, concern about reliance on AMP
C		Conseq. Costs	Major	Possible	High	
D		Spec. Cons.	Moderate	Possible	Moderately High	
E	Source water quality (source term) worse than expected and causes unacceptable water quality conditions **on site**	Env. Imp.	Moderate	Unlikely	Moderate	
F		Legal Obl.	Major	Possible	High	This could apply to the preceding mode if there were W2 standards in place
G		Conseq. Costs	Moderate	Possible	Moderately High	
H	Precipitation higher than expected resulting in failure of water conveyance structures because structures are underdesigned	Conseq. Costs	Moderate	Possible	Moderately High	assuming 200 yr flood design, note - sensitivity analysis for precip on water quality,
I	Localized precip > regional => less dilution in downstream in downstream environment resulting in unacceptable water quality conditions downstream of site	Env. Imp.	Minor	Likely	Moderate	
2	**Cover Performance**					
A	Covers don't perform as designed re: infiltration resulting in unacceptable water quality conditions **downstream of site**	Env. Imp.	Moderate	Unlikely	Moderate	Assuming AMP and monitoring will respond to any defects
B		Conseq. Costs	Major	Unlikely	Moderately High	
3	**Geotechnical Stability**					
A	Geotechnical failure of any waste facility (slope stability) resulting in waste material exposure to water leading to unacceptable water quality conditions downstream of site	Env. Imp.	Moderate	Possible	Moderately High	this is a result of there being permafrost considerations under some facilities, and uncertainty associated. Could result from differential settlement of pockets of more moist materials
B		Conseq. Costs	Moderate	Possible	Moderately High	
C	Geotechnical failure of any waste facility (differential settlement) resulting in rupture of cover and waste material exposure to water leading to unacceptable water quality conditions downstream of site	Env. Imp.	Moderate	Likely	Moderately High	Could result from differential settlement of pockets of high moisture materials. Could be moderated by waste mgmt practices limiting wet waste in waste dumps
D	Geotechnical failure of any waste facility (slope stability) resulting in debris dam, breaching, mobilizing materials and pulse of water into Main Pit, and sediments/tailings leaving pit, leading to unacceptable water quality conditions downstream of site	Env. Imp.	Minor	Possible	Moderate	
4	**Conveyance Structures**					
A	Undiverted runoff upstream of waste mgmt facilities leads to runon water, extra infiltration, leading to ongoing maintenance costs	Conseq. Costs	Moderate	Unlikely	Moderate	Need to do landscape design carefully to avoid this failure mode
B	Undiverted runoff upstream of waste mgmt facilities leads to excessive infiltration into upgradient base of dump, resulting in higher flows of poor quality water and unacceptable water quality conditions downstream	Env. Imp.	Moderate	Unlikely	Moderate	
5	**Administrative**					
A	Failure to implement AMP, resulting in unacceptable water quality conditions downstream	Env. Imp.	Critical	Possible	High	
B	Failure to design an appropriate AMP, resulting in unacceptable water quality conditions downstream	Env. Imp.	Major	Unlikely	Moderately High	Important to recognize that AMP is more than just monitoring - but careful identification of potential issues, thresholds and appropriate responses. AMP not just an add-on. Needs to be critical component of closure plan at same detail as rest of plan
C	Departure from design of engineered structures, resulting in unacceptable water quality conditions downstream	Env. Imp.	Major	Possible	High	

	Scenario 2: Hybrid Source Control/Passive Treatment Focus	Consequence		Likelihood	Risk Rating	NOTES
		Type	Severity	Probability	Descriptive	
1	**Bioreactors**					
A	Bioreactors don't perform as designed - overwhelmed, freeze, resulting in unacceptable water quality conditions downstream	Env. Imp.	Moderate	Almost Certain	High	Assumption: effective AMP in place Notes: cryo-concentration in seeps, ice cleaner, residual seeps higher concentration Leslie: make sure that any supporting work here has data - not just stories that they work -i.e. Andre Sobolewski's work at G900 didn't work, but MPERG report still says it does.
2	**Wetlands**					
A	Wetlands don't perform as designed - overwhelmed, freeze, resulting in unacceptable water quality conditions downstream	Env. Imp.	Moderate	Almost Certain	High	
B	High flow blow out wetland, causing damage and maintenance requirements, assuming high flows designed to bypass	Env. Imp.	Moderate	Unlikely	Moderate	Assumption: peak flows not treated by wetlands - need to understand the implications of this during freshet and also during peak flow events. Wetlands are not designed nor capable of treating peak flows, so this is a significant red-flag for planning - needs very careful consideration.
C		Conseq. Costs	Moderate	Unlikely	Moderate	Need to understand the implications of this during freshet and also during peak flow events.
3	**Pit Lake Treatment**					
A	Non-Flow through Pit:					
i	Non-flow through Area 2 Pit treatment compromised because of diversion ditch failure, resulting in flow through condition	Env. Imp.	Minor	Possible	Moderate	Assuming pit water quality has moderate initial contamination level - make sure this is covered in AMP. What if WQ in Area 2 pit were higher than anticipated?
ii	Non-flow through pit treatment does not perform	Env. Imp.	Minor	Possible	Moderate	
iii	Pit Wall Failure in Area 2 results in wave of water released from pit causing damage to downstream facilities and tailings mobilization from bottom of Area 2 pit	Conseq. Costs	Major	Unlikely	Moderately High	Resolution would be difficult - would mean appropriate sizing of the spillway, locating of facilities downgradient
B	Flow through Pit:					
	Flow through Pit treatment does not perform as expected	Env. Imp.	Minor	Unlikely	Low	Assume treatment expectations consider flow through condition and limitations
	Flow through Pit source term underestimated, resulting in higher than expected loading from pit and unacceptable water quality results downstream	Env. Imp.	Moderate	Unlikely	Moderate	
	Pit Wall Failure in Area 2 results in wave of water released from pit causing damage to downstream facilities and tailings mobilization from bottom of Area 2 pit	Conseq. Costs	Major	Very Unlikely	Moderate	In the flow through pit condition, the downstream channels and facilities would be designed for hitgher flows, so likelihood lower than in the non-flow through condition.
4	**Cover Performance**					
A	Does not achieved expected infiltration reduction, leads to increased loading and unacceptable downstream WQ effects	Env. Imp.	Moderate	Possible	Moderately High	
B	Erosion leads to increased infiltration and unacceptable downstream WQ effects	Env. Imp.	Moderate	Unlikely	Moderate	Risks different for DSTSF than for other facilities, potential effects of erosion still need to be considered in design, maintenance costing, etc.
5	**Conveyance Structures**					
A	Undiverted runoff upstream of waste mgmt facilities leads to runon water, extra infiltration, leading to ongoing maintenance costs	Conseq. Costs	Minor	Unlikely	Low	Need to do landscape design carefully to avoid this failure mode
B	Undiverted runoff upstream of waste mgmt facilities leads to excessive infiltration into upgradient base of dump, resulting in higher flows of poor quality water and unacceptable water quality conditions downstream	Env. Imp.	Moderate	Possible	Moderately High	
6	**Administrative**					
A	General failure to maintain site requirements as required - passive treatment, cover maintenance, etc.	Env. Imp.	Major	Possible	High	

	Scenario 3: Treatment Focus	Consequence		Likelihood	Risk Rating	NOTES
		Type	Severity	Probability	Descriptive	
	Assumption is that this option needs redesign compared with existing collection/treatment system in place - many risks could be addressed through this redesign, or batch treatment, etc.					
1	**Collection Systems**					
A	Tailings seepage collection systems inadequate, leading to unacceptable WQ downstream	Env. Imp.	Moderate	Possible	Moderately High	design, size, location, construction, operation - all contributors to the potential issue, these need to be thought through more for the mitigation
B		Conseq. Costs	Major	Possible	High	
C	SWD toe seepage collection systems inadequate, leading to unacceptable WQ downstream	Env. Imp.	Minor	Likely	Moderate	minor because pit is downstream
D		Conseq. Costs	Moderate	Likely	Moderately High	Feasiblilty of this collection system questionable - due to ice-rich area and deformations, and no clear segregation from valley flows. Mitigation might be to avoid collection system altogher and focus on treatment of full W15 flow in pit.
E	Collection of cleaner runoff in inadequate, leading to mixing with water requiring treatment and increased treatment costs	Conseq. Costs	Minor	Likely	Moderate	
2	**Cover Performance**					
A	Does not achieved expected infiltration reduction, leads to increased loading and unacceptable downstream WQ effects	Env. Imp.	Moderate	Possible	Moderately High	Assumption that these Option 3 covers are thinner than Option 2
B	Erosion leads to increased infiltration and unacceptable downstream WQ effects	Env. Imp.	Moderate	Possible	Moderately High	risks different for DSTSF than for other facilities, potential effects of erosion still need to be considered in design, maintenance costing, etc.
C	Undiverted runoff upstream of waste mgmt facilities leads to runon water, extra infiltration, leading to ongoing maintenance costs	Conseq. Costs	Moderate	Unlikely	Moderate	Need to do landscape design carefully to avoid this failure mode
D	Undiverted runoff upstream of waste mgmt facilities leads to excessive infiltration into upgradient base of dump, resulting in higher flows of poor quality water and unacceptable water quality conditions downstream	Env. Imp.	Moderate	Possible	Moderately High	
3	**Dam - assume reduced height**					**may want to consider removing top level at least - man/made materials**
A	Seismic or extreme flood event larger than design leads to dam failure, resulting in surge of water and solids into Minto Creek	Env. Imp.	Moderate	Very Unlikely	Low	
B	Dam maintenance requirements not met, resulting in failure and surge of water and solids into Minto Creek	Env. Imp.	Moderate	Very Unlikely	Low	assumes design with maintenance requirements
4	**Treatment Plant (plant, any byproduct, and storage capacity)**	**Env. Imp.**	**Very Low**	**Very Unlikely**	**Low**	
A	Flow rates exceed plant/surge capacity, resulting in unnacceptable water quality downstream	Env. Imp.	Moderate	Possible	Moderately High	Mitigation : increase surge capacity and/or operate surge volumes better - depending on why surge capacity was overwhelmed
B		Conseq. Costs	Major	Unlikely	Moderately High	assume worst case - plant/surge exceeded because not sufficient
C	Contaminant loading exceeds treatment capacity, resulting in unnacceptable water quality downstream	Env. Imp.	Moderate	Unlikely	Moderate	Function of geochemical source term identification
D		Conseq. Costs	Major	Unlikely	Moderately High	
E	Treatment technology ineffective for contaminants of concern, resulting in unnacceptable water quality downstream	Env. Imp.	Moderate	Very Unlikely	Low	
F		Conseq. Costs	Major	Very Unlikely	Moderate	
G	Inadequate capacity for storage of byproducts, leads to costs for removal off site	Conseq. Costs	Moderate	Very Unlikely	Low	
5	**Administrative**	**Env. Imp.**	**Very Low**	**Very Unlikely**	**Low**	
A	General failure to maintain site requirements as required - collection/conveyance, active treatment, cover maintenance, etc.	Env. Imp.	Major	Possible	High	

MINTO MINE CLOSURE – FAILURE MODES AND EFFECTS ASSESSMENT

2014 WORKSHOP REPORT

FINAL REPORT

December 2014

Prepared for:

MINTO EXPLORATIONS LTD.

ACCESS

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1 INTRODUCTION

In advance of the submission of a Reclamation and Closure Plan (RCP) for the Phase IV mine plan at the Minto Mine, Minto Explorations Ltd. (Minto) conducted a multi-stakeholder Failure Modes and Effects Assessment (FMEA) on a suite of example mine closure scenarios. The workshop was held in Whitehorse in January of 2013, and involved participants representing Minto, Selkirk First Nation (SFN), Yukon Government- Energy Mines and Resources (YG-EMR), and the Yukon Water Board (YWB).

On behalf of Minto, Access Consulting Group has developed and recently submitted an updated Reclamation and Closure Plan (RCP v5.1, August 2014) to support permit amendment applications for the Phase V/VI Expansion mine plan.

Minto hosted a multi-stakeholder FMEA workshop (in two parts) for the Phase V/VI RCP. The first session in Vancouver on August 27 and 28, 2014. A second, supplementary FMEA workshop was held in Whitehorse on October 9 and 10 with a smaller subset of the original workshop group to address outstanding mine components and closure aspects that had not been addressed in the first workshop.

As in January 2013, the FMEA used predefined consequence categories, severity descriptors and likelihood terminology to determine where the residual risk associated with the various mine components ranked on the risk matrix (from Low to Very High). The risk rating tools used in the FMEA workshop(s) is presented in Appendix A.

The terminology adopted for the risk rating tools was modified slightly from the terminology used in the January 2013 FMEA workshop to reflect feedback from a broad range of workshop participants during a pre-workshop teleconference.

The FMEA workshop methodology and agenda, which was also refined during a pre-workshop teleconference, was distributed to workshop participants in advance of the workshop (Appendix B).

This report provides a description of the FMEA objectives and scope, and a summary of the workshop and the outcomes. It should be noted that some of the facilities initially identified for inclusion in the FMEA workshop were, upon further discussion with the broader workshop group, deemed to be relatively low risk and therefore dropped from detailed discussion and formal risk ranking exercise.

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2 WORKSHOP OBJECTIVES, SCOPE AND APPROACH

The overall objective of the FMEA workshop was to evaluate the residual risks that would remain after implementation of the RCP v5.1.

The FMEA covered the entire Minto mine site and mainly focused on a time frame during Post Closure II as it is described in the RCP v5.1, when all reclamation activities are completed and the site has entered into a phase of primarily monitoring and maintenance.

The Closure FMEA used an approach similar to that utilized in the Preliminary FMEA in January 2013, in which specific combinations of failure modes and resulting effects were rated by participants. The failure scenarios generated in 2013 were revisited at the beginning of this workshop to preserve considerations which were still relevant to the Phase V/VI RCP. Appendix C-1 presents the relevant risk scenarios that were carried over from the January 2013 FMEA workshop. One of the first tasks of the FMEA workshop participants was to determine, as a group, to which facilities the identified risk scenarios would apply.

The FMEA was conducted both on a "Facility" basis (separate risk register for each facility) and on topics that are appropriately addressed on a "Site-Wide" basis. These are generally reflective of the reclamation and closure measures presented in RCP Section 7, and an initial numbered starting list of topic areas included:

Facilities:

1. Underground Workings – subsidence, hydrologic
2. Open Pits
3. Dry Stack Tailings Storage Facility
4. Main Dam
5. Mill Valley Fill Extension
6. Waste Rock Dumps
7. Overburden Dumps
8. Ore Stockpiles and Pads
9. Mine Infrastructure

Site-Wide Topics:

10. Source Control – Waste Covers
11. Water Treatment
12. Water Conveyance
13. Site Access
14. Administrative
15. "Domino Effect" (added by consensus during August workshop)

The columns proposed for each of the above risk registers were:

- Category of failure (as appropriate for facility/topic)
  - water management
  - physical stability

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  - chemical stability
  - administration
- Scenario combining failure mode and effect
- Consequence type
- Consequence severity
- Likelihood of occurrence
- Risk Rating
- Notes/ Mitigations

The risk rating tools utilized during the workshop are included in Appendix A.

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3 OVERVIEW OF FMEA WORKSHOP

3.1 FMEA PROCESS

The FMEA was carried out using a consequence-likelihood method, utilizing three risk rating tools (the consequence-severity matrix, the likelihood chart, and the risk matrix) located in Appendix A. A draft agenda and methodology was distributed to all participants for consideration and comment. One week prior to the workshop, a conference call was held which included representatives from Minto, SRK, SFN, and Access, to discuss and refine the methodology so as to maximize the time for risk ratings during the workshop. During this call it was determined that the results of the January 2013 FMEA workshop should be somehow incorporated so as to not lose the information gained from that process. It was agreed that Access would compile the scenarios into a starting table to evaluate at the beginning of the workshop.

The final agenda/methodology document (revised through advance participant discussion) is included here for reference as Appendix B.

3.2 WORKSHOP STRUCTURE

The workshop began with a review of the approach, and the agreement that the timeframe of the FMEA was to examine risks to the site during the Post Closure II period when all site reclamation work was complete and the site was in a state of monitoring and maintenance. The group agreed to modify the consequence-severity matrix such that a single fatality would result in a "Critical" severity rating, as opposed to a rating of "Major".

A discussion was held regarding the risk matrix and how the process is evaluating annual residual risk. Concern was raised that it was important to consider longer term views. It was noted that there is a distinct difference in significance for a 200-year event to First Nations and to a mining proponent or consultant. It was agreed to move forward on the basis of evaluating annual residual risk, but to acknowledge concern of longer term risks as appropriate and to flag issues for further discussion.

As suggested in the discussions held during the conference call prior to the workshop, the 2013 FMEA risk registry was consolidated onto a single worksheet by Access and given preliminary categories for filing under the various facilities and site wide topics. The group reviewed this preliminary allocation and adjusted if required. The final allocation table is provided in Appendix C. The 2013 risk registry results were then migrated to the appropriate 2014 FMEA risk register with the understanding that the wording of each previous failure mode would be reworked to reflect the current RCP if/as appropriate.

3.2.1 Risk Rating Process

The risk ratings were developed by the group. For each scenario that was rated, potential risks were identified, recorded, and taken through a facilitated procedure using the consequence-severity and likelihood tools to reach a consensus risk rating. The risk ratings were recorded in a risk register spreadsheet that was projected on a screen for participants to refer to and provide feedback on during the meeting, and the resultant risk IDs were placed on a wall matrix and photographed once the topic was complete.

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4 FMEA RESULTS

4.1 AUGUST WORKSHOP

The first FMEA workshop was held at SRK's office in Vancouver, BC on August 27 and 28, 2014. The workshop was facilitated by Dr. Dirk Van Zyl (Chair of Mining and the Environment at the Norman B. Keevil Institute of Mining Engineering, University of British Columbia). The two-day workshop included participation by representatives of Minto, SFN, and YG-EMR. Representatives from Norwest Corporation (in its third party review capacity on geotechnical subjects at Minto Mine, on behalf of Minto and SFN jointly) participated on the second day only (August 28).

The participants are listed in Table 1.

Table 4-1 August FMEA Workshop Participants

Name	Company	Days Attending
Jim Theriault	Access	August 27/28
Ken Boldt	Access	August 27/28
Scott Keesey	Access	August 27/28
Dirk Van Zyl	SRK	August 27/28
Peter Mikes	SRK	August 27/28
Ryan Herbert	Minto	August 27/28
Jennie Gjertsen	Minto	August 27/28
Erin Dowd	YG Mineral Resources	August 27/28
Bill Slater	BSEC	August 27/28
William Sydney	SFN	August 27/28
Cord Hamilton	Northland Earth & Water	August 27/28
Jim Kuipers	KA/SFN	August 27/28
Debbie Trudeau	SFN	August 27/28
Pooya Mohseni	Capstone/ Minto	August 27/28
Dylan MacGregor	SRK	August 27/28
Cam Scott	SRK	August 28
Richard Dawson	Norwest	August 28
David Sego	Norwest	August 28

It was apparent to the participants early in the workshop that the initial list of topics could not be addressed adequately in the two-day allotted timeframe. It was therefore decided by the group that certain facilities and topics were not of sufficient consequence to warrant formal rate by the FMEA process. These topics were removed from the agenda, and included: (1) Underground Workings; (4) Overburden Dumps; (5) Ore Stockpile Pads; (9) Mine Infrastructure and (13) Site Access.

A discussion regarding where to place scenarios that involved a number of facilities and 'cascading' effects led to the creation of a new site-wide topic called "Domino Effect (#15).

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Key remaining topics were prioritized for completion in this workshop, and it was agreed that the remaining topics would be postponed to be handled at a future time. Access agreed to construct "strawmen" scenarios for these facilities and topics to present to the larger group to expedite the follow-up session.

The topics rated in the August workshop included:

August 27:

1. Area 2 Pit
9. Source Control – Waste Covers
10. Water Conveyance

August 28:

2/3/6. Main Pit/Main Pit Dump/Main Dam and Spillway

7/8. Dry Stack Tailings Storage Facility and Mill Valley Fill Extension

Topics that were deferred to the supplementary FMEA workshop included:

Facilities:

2. Open Pits (Minto North and Ridgetop South)
3. Waste Rock Dumps (Main Waste Dump, South West Dump, Ridgetop Waste Dump)

Site-Wide Topics:

11. Water Treatment
14. Administrative
15. Domino Effect

4.2 OCTOBER WORKSHOP

The supplementary FMEA workshop was conducted over two half-day sessions to accommodate scheduling challenges and ensure that key participants with specialist expertise were present for the appropriate discussions. The workshop was jointly facilitated by Dylan MacGregor and Scott Keesey and the same FMEA protocols and risk rating tools that were used in the August workshop were used once again.

As agreed to during the August workshop, participation in the second FMEA workshop was limited to those who had participated in the first workshop. The one notable exception was Steve Januszewski (SJCI), an independent consultant who is intimately familiar with the Minto Site and participated on the 2013 FMEA process. Steve attended on behalf of YG EMR Mineral Resources.

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Table 4-2 October FMEA Workshop Participants

Name	Company	Days Attending
Jim Theriault	Access	October 9/10
Scott Keesey	Access	October 9/10
Dylan MacGregor	SRK	October 9/10
Jennie Gjertsen	Minto	October 9/10
Pooya Mohseni	Capstone/Minto	October 9
William Sydney	SFN	October 9
Jim Kuipers	KA/SFN	October 9/10
Debbie Trudeau	SFN	October 9/10
Steve Januszewski	SJCI/YG Mineral Resource	October 9/10
Cord Hamilton	Northland Earth and Water	October 10

As proposed during the August workshop, Access pre-populated the Risk Scenarios with relevant examples from the August workshop and also proposed some additional risk scenarios for the wider group to consider. This provided an efficient starting point and all Scenarios were subsequently vetted and expanded upon during the FMEA workshop.

The afternoon of October 9th was devoted to risk ranking scenarios from the remaining WRDs (Main Waste Dump, Southwest Dump and Ridgetop Waste Dump). The risk rankings from the Main Pit Dump (completed during the August FMEA workshop) were reviewed to help recalibrate the group and reacquaint everyone with the FMEA process. Discussions were also expanded to include backfill dumps (which had previously been removed from planned discussions) as it was considered important to address a potential SFN concern/perception that the proposed overburden backfill activities could be seen as wasting good reclamation materials.

The October 10th session addressed the remaining open pits (Minto North, Ridgetop North), Water Treatment, Administration and "Domino Effect". With respect to Water Treatment, it was recognized that reclamation research into passive treatment is ongoing and evolving. It was agreed that the Reclamation Research Plan needs to advance further in order to better evaluate reasonable and practicably treatment technologies and there is a need to better define water quality objectives before Water Treatment can be properly evaluated by the FMEA process. The group ultimately decided to not rank Water Treatment scenarios but rather flag this topic as significant and consider addressing residual risks associated with Water Treatment using a different process in the near future.

The results of the FMEA risk ratings completed during the October workshop are presented in Appendix C-3.

The workshop participants also identified a number of "parking lot" issues/concerns that could not be addressed by the FMEA process but which require further consideration. The key issues/concerns raised included:

- Current closure plan is deficient with respect to showing final reclaimed facilities, toes of re-graded slopes and location of secondary and tertiary water conveyance;

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- More information requested regarding the status of the Reclamation Research Plan and the Main Waste Dump revegetation trials;

- Trafficability layer is required over the Ridgetop North Pit tailings backfill whereas costing only allows for 0.5m of overburden;

- Need to advance the discussion/determination of what constitutes "reasonable and practicable" passive treatment, establish protocols and revisit the options evaluation;

- SFN reiterated their concern that the consequence category of "Community/Media/Reputation" is biased because SFN are lumped together with groups having other interests and perspective; and

- The current closure plan does not sufficiently address signage and access control. There is a need to retain institutional controls and maintain signage in perpetuity.

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5 SUMMARY

The risk registers developed during the workshop are provided in Appendix C. A complete summary of the risk scenarios considered during the August and October FMEA workshops is presented on a single table in Appendix D.

The August workshop rated breach of the Main Dam due to permafrost thaw, settling of the dam leading to tailings release, and general failure to conduct preventative maintenance and corrective actions leading to system failures, as the largest perceived residual risks to the site at closure. The lack of inclusion of long-term operation and maintenance of the site was identified as an important issue to be addressed, as was further assessment of permafrost thaw as it pertains to the closure design, in particular for the Main Dam and associated structures. A number of scenarios were not rated, but identified further investigation was required. Other scenarios were given 'provisional' ratings, which were based on certain assumptions which require confirmation/follow up.

The October workshop, which focused primarily on WRDs and open-pits that had not been addressed during the August workshop, generally found the risk scenarios evaluated to contain relatively low residual risk (i.e., Low to Moderate) and primary mitigative actions identified included implementation of an effective AMP and minor modifications to closure configuration (e.g. regrading to maintain ponds away from spill points).

A summary of all risk scenarios evaluated in the August and October FMEA workshops is presented in Appendix D. The highest overall risks (i.e., High and Very High) identified during the FMEA workshops were associated with Administration (Category 14), Water Conveyance (12) and Waste Rock Dumps (6), with Administrative Failures representing the largest perceived residual closure risk. A recurring theme for mitigating residual risk included the implementation of an effective AMP and long-term care and maintenance program.

The results of the two FMEA workshop sessions, combined with formal RCP review comments will be useful in evaluating the need for modifications or improvements to the RCP in advance of completion of the Phase V/VI permitting process.

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6 REFERENCES

Access Consulting Group. 2014. Minto Mine Phase V/VI Expansion, Reclamation and Closure Plan, Revision 5.1. Prepared for Capstone Mining Corp., August 2014.

Access Consulting Group. 2013. Minto Mine Phase IV Reclamation and Closure Plan, Revision 4.0. Prepared for Capstone Mining Corp., September 2013.

APPENDIX A

RISK RATING TOOLS

Table 1. Consequence-Severity Matrix

	Severity Descriptors				
Consequence Categories	**Very Low**	**Minor**	**Moderate**	**Major**	**Critical**
1. Environmental Impact	No impact.	Minor localized or short-term impacts.	Significant impact on valued ecosystem component.	Significant impact on valued ecosystem component and medium-term impairment of ecosystem function.	Serious long-term impairment of ecosystem function.
2. Traditional Use	Some disturbance but no impact to traditional land use.	Minor or perceived impact to traditional land use.	Some mitigable impact to traditional land use.	Significant temporary impact to traditional land use.	Significant permanent impact on traditional land use.
3. Regulatory and Legal	Informal advice from a regulatory agency.	Technical/Administrative non-compliance with permit, approval or regulatory requirement. Warning letter issued.	Breach of regulations, permits, or approvals (e.g. 1 day violation of discharge limits). Order or direction issued.	Substantive breach of regulations, permits or approvals (e.g. multi-day violation of discharge limits). Prosecution.	Major breach of regulation – wilful violation. Court order issued.
4. Consequence Costs	< $100,000	$100,000 - $500,000	$ 500,000 - $2.5 Million	$2.5-$10 Million	>$10 Million
5. Community/ Media/ Reputation	Local concerns, but no local complaints or adverse press coverage.	Public concern restricted to local complaints or local adverse press coverage.	Heightened concern by local community, criticism by NGOs or adverse local /regional media attention.	Significant adverse national public, NGO or media attention.	Serious public outcry/demonstrations or adverse International NGO attention or media coverage.
6. Human Health and Safety	Low-level short-term subjective symptoms. No measurable physical effect. No medical treatment.	Objective but reversible disability/impairment and /or medical treatment injuries requiring hospitalization.	Moderate irreversible disability or impairment to one or more people.	Severe irreversible disability or impairment to one or more people.	Single fatality or multiple fatalities.

Table 2. Likelihood Terminology

Likelihood	Frequency Descriptor 1	Frequency Descriptor 2
Almost Certain	Happens often	High frequency (more than once every 5 years)
Likely	Could easily happen	Event does occur, has a history, once every 15 years
Possible	Could happen and has happened elsewhere	Occurs once every 40 years
Unlikely	Hasn't happened yet but could	Occurs once every 200 years
Very Unlikely	Conceivable, but only in extreme circumstances	Occurs once every 1000 years

Table 3. Risk Matrix

Likelihood	Consequence Severity				
	Very Low	**Minor**	**Moderate**	**Major**	**Critical**
Almost Certain	**Moderate**	**Moderately High**	**High**	**Very High**	**Very High**
Likely	**Moderate**	**Moderate**	**Moderately High**	**High**	**Very High**
Possible	**Low**	**Moderate**	**Moderately High**	**High**	**High**
Unlikely	**Low**	**Low**	**Moderate**	**Moderately High**	**Moderately High**
Very Unlikely	**Low**	**Low**	**Low**	**Moderate**	**Moderately High**

APPENDIX B

MINTO MINE FMEA METHODOLOGY AND AGENDA

Minto Explorations Ltd.

Minto Mine Phase V/VI Expansion
Methodology for Closure FMEA Workshop

1 Introduction

In advance of the submission of a Reclamation and Closure Plan (RCP) for the Phase IV mine plan at the Minto Mine, Minto conducted a multi-stakeholder Failure Modes and Effects Assessment (FMEA) on a suite of example mine closure scenarios. The workshop was held in Whitehorse in January of 2013, and involved participants representing Minto, Selkirk First Nation (SFN), Yukon Government- Energy Mines and Resources (YG-EMR), and the Yukon Water Board (YWB). It was recognized that the January 2013 FMEA was preliminary in nature, and that a follow-up FMEA would be appropriate once a set of closure options and measures was established. The Phase IV RCP (ACG 2013) incorporated outcomes from the January 2013 FMEA, and was completed and submitted to regulators in September 2013. The risk register from the January 2013 FMEA was appended to the Phase IV RCP; the follow-up FMEA was not undertaken prior to completion of the Phase IV RCP. In meetings between Minto and SFN since the January 2013 FMEA, the merits of completing the exercise have been discussed numerous times.

Minto has recently submitted an updated Reclamation and Closure Plan (RCP v5.1, August 2014) to support permit amendment applications for the Phase V/VI Expansion mine plan. Minto will host a multi-stakeholder FMEA for the Phase V/VI RCP in Vancouver on August 27 and 28, 2014. As in January 2013, the workshop will be facilitated by Dr. Dirk Van Zyl (Chair of Mining and the Environment at the Norman B. Keevil Institute of Mining Engineering, University of British Columbia). The two-day workshop will include participation by representatives of Minto, SFN, and YG-EMR. Representatives from Norwest Corporation (in its third party review capacity on geotechnical subjects at Minto Mine, on behalf of Minto and SFN jointly) will participate on the second day only (August 28); geotechnical considerations will be discussed most substantially on the second day (August 28) given key participant availability. Follow up sessions will be conducted as and if required.

This document provides a description of the FMEA objectives and scope, and the proposed approach to the workshop. The approach will be reviewed and finalized after input from participants via conference call in advance of the workshop.

2 Workshop Objectives

The overall objective of the FMEA workshop is to evaluate the residual risks that would remain after implementation of the RCP v5.1.

3 Scope and Approach

The FMEA will cover RCP v5.1 and related design elements for the proposed closure measures. Familiarity with the RCP v5.1 document for all participants will be critical to a meaningful

contribution to the FMEA workshop. Familiarity with facility design reports referenced in the RCP will also be necessary. Access to all of these reports will be provided via FTP site.

The Closure FMEA will be carried out using an approach similar to that utilized in the Preliminary FMEA in January 2013, in which specific combinations of failure modes and resulting effects were rated by participants. Further details on the proposed method are provided in Section 4.

A series of risk registers will be developed (in table format) during the workshop. The FMEA will be conducted both on a "Facility" basis (separate risk register for each facility) and on topics that are appropriately addressed on a "Site-Wide" basis. These are generally reflective of the reclamation and closure measures presented in RCP Section 7, and will include:

Facilities:

- Underground Workings – subsidence, hydrologic
- Open Pits
- Dry Stack Tailings Storage Facility
- Main Dam
- Mill Valley Fill Extension
- Waste Rock Dumps
- Overburden Dumps
- Ore Stockpiles and Pads
- Mine Infrastructure

Site-Wide Topics:

- Source Control – Waste Covers
- Water Treatment
- Water Conveyance
- Site Access
- Administrative

The columns for each of the above risk registers will be:

- Category of failure (as appropriate for facility/topic)
  - water management
  - physical stability
  - chemical stability
  - administration
- Scenario combining failure mode and effect
- Consequence type
- Consequence severity
- Likelihood of occurrence
- Risk Rating
- Notes/ Mitigations

4 Risk Rating Tools

This section presents the tools that form the basis of the risk rating method proposed for the workshop. The tools include two tables and a risk matrix; draft versions of the tools are shown in Tables 1 through 3.

Table 1 presents six categories of consequences that will be considered along with severity ratings ranging from “Very Low” to “Critical”. For each category, the table includes narrative descriptions of the types of negative outcomes that would be typical for each severity rating; these descriptions will be used for reference during the workshop to help participants determine the appropriate severity rating to be assigned to a scenario.

Table 1. Consequence-Severity Matrix

	Severity Descriptors				
Consequence Categories	**Very Low**	**Minor**	**Moderate**	**Major**	**Critical**
1. Environmental Impact	No impact.	Minor localized or short-term impacts.	Significant impact on valued ecosystem component.	Significant impact on valued ecosystem component and medium-term impairment of ecosystem function.	Serious long-term impairment of ecosystem function.
2. Traditional Use	Some disturbance but no impact to traditional land use.	Minor or perceived impact to traditional land use.	Some mitigable impact to traditional land use.	Significant temporary impact to traditional land use.	Significant permanent impact on traditional land use.
3. Regulatory and Legal	Informal advice from a regulatory agency.	Technical/Administrative non-compliance with permit, approval or regulatory requirement. Warning letter issued.	Breach of regulations, permits, or approvals (e.g. 1 day violation of discharge limits). Order or direction issued.	Substantive breach of regulations, permits or approvals (e.g. multi-day violation of discharge limits). Prosecution.	Major breach of regulation – wilful violation. Court order issued.
4. Consequence Costs	< $100,000	$100,000 - $500,000	$ 500,000 - $2.5 Million	$2.5-$10 Million	>$10 Million
5. Community/ Media/ Reputation	Local concerns, but no local complaints or adverse press coverage.	Public concern restricted to local complaints or local adverse press coverage.	Heightened concern by local community, criticism by NGOs or adverse local /regional media attention.	Significant adverse national public, NGO or media attention.	Serious public outcry/demonstrations or adverse International NGO attention or media coverage.
6. Human Health and Safety	Low-level short-term subjective symptoms. No measurable physical effect. No medical treatment.	Objective but reversible disability/impairment and /or medical treatment injuries requiring hospitalization.	Moderate irreversible disability or impairment to one or more people.	Severe irreversible disability or impairment to one or more people.	Single fatality or multiple fatalities.

Table 2 presents descriptors the will be used to aid participants in assigning a 'Likelihood' rating for each scenario. The 'Likelihood' rating will be assigned to reflect the participants' view on the probability both that the failure mode will occur and that the effect will result- a series of terms used to define the likelihood that a consequence (from the previous chart) will be realized. The 'Likelihood Terminology' table consists of one column containing likelihood ratings that range from 'Very Unlikely' to 'Almost Certain', along with four other columns which give examples to guide the selection of the appropriate rating.

Table 2. Likelihood Terminology

Likelihood	Frequency Descriptor 1	Frequency Descriptor 2
Almost Certain	Happens often	High frequency (more than once every 5 years)
Likely	Could easily happen	Event does occur, has a history, once every 15 years
Possible	Could happen and has happened elsewhere	Occurs once every 40 years
Unlikely	Hasn't happened yet but could	Occurs once every 200 years
Very Unlikely	Conceivable, but only in extreme circumstances	Occurs once every 1000 years

Table 3 presents the 'Risk Matrix' which assigns each combination of severity (Table 2) and likelihood (Table 1) a risk rating. This matrix will be used in the workshop to supplement the recording of the results in the risk registry.

Table 3. Risk Matrix

Likelihood	Consequence Severity				
	Very Low	Minor	Moderate	Major	Critical
Almost Certain	Moderate	Moderately High	High	Very High	Very High
Likely	Moderate	Moderate	Moderately High	High	Very High
Possible	Low	Moderate	Moderately High	High	High
Unlikely	Low	Low	Moderate	Moderately High	Moderately High
Very Unlikely	Low	Low	Low	Moderate	Moderately High

5 Workshop Report

Following the workshop, Minto will compile a report summarizing the workshop methods and outcomes, and will circulate the draft report to the workshop participants for comment. The following is a draft outline for the report.

- Introduction
- Workshop objectives
- The boundaries of the FMEA
- A description of the workshop method
- Products of the workshop
- Comments on draft report by workshop participants

6 References

Access Consulting Group. 2014. Minto Mine Phase V/VI Expansion, Reclamation and Closure Plan, Revision 5.1. Prepared for Capstone Mining Corp., August 2014.

Access Consulting Group. 2013. Minto Mine Phase IV Reclamation and Closure Plan, Revision 4.0. Prepared for Capstone Mining Corp., September 2013.

APPENDIX C

RISK REGISTERS

Original Scenario		2013 Minto Phase IV Closure FMEA - Scenarios Relevant to Phase V/VI	Destination	Consequence		Likelihood	Risk Rating	NOTES
				Type	Severity	Probability	Descriptive	
	1	**Water Management**						
	H	Precipitation higher than expected resulting in failure of water conveyance structures because structures are underdesigned	2,3,6,7,8,10,12,15	Conseq. Costs	Moderate	Possible	Moderately High	assuming 200 yr flood design, note - sensitivity analysis for precip on water quality,
	I	Localized precip > regional => less dilution in downstream in downstream environment resulting in unacceptable water quality conditions downstream of site	10,11	Env. Imp.	Minor	Likely	Moderate	
	A	Undiverted runoff upstream of waste mgmt facilities leads to runon water, extra infiltration, leading to ongoing maintenance costs	2W, 3W, 4W, 6W,7W, 8W,12,15	Conseq. Costs	Moderate	Unlikely	Moderate	Need to do landscape design carefully to avoid this failure mode
	B	Undiverted runoff upstream of waste mgmt facilities leads to excessive infiltration into upgradient base of dump, resulting in higher flows of poor quality water and unacceptable water quality conditions downstream	3W, 4W, 7W, 8W,12,15	Env. Imp.	Moderate	Unlikely	Moderate	
	1	Failure of existing TDD leads to erosion/channeling and mobilizing materials off faclility during operations		Env. Imp.	Minor	Almost Certain	Moderately High	
	2	Failure of diversion leads to erosion/channeling through cover and into tailings, mobilizing up to 50 tonnes of tailings all the way to Lower Minto Creek during closure	7W, 12, 15	Env. Imp.	Major	Unlikely	Moderately High	FMEA process for Phase IV should inform next version of RCP - re: inspection frequency and what inspection programs/ instrumentation should look like. Assuming annual inspections (1st 5 years? - Scott to check) Lower Minto Creek is a relatively small and relatively unproductive ecosystem
	A	Tailings seepage collection systems inadequate, leading to unacceptable WQ downstream	2W, 6W, 7W,12	Env. Imp.	Moderate	Possible	Moderately High	design, size, location, construction, operation - all contributors to the potential issue, these need to be thought through more for the mitigation
	B			Conseq. Costs	Major	Possible	High	
	C		3W, 12	Env. Imp.	Minor	Likely	Moderate	minor because pit is downstream
	D	SWD toe seepage collection systems inadequate, leading to unacceptable WQ downstream		Conseq. Costs	Moderate	Likely	Moderately High	Feasiblilty of this collection system questionable - due to ice-rich area and deformations, and no clear segregation from valley flows. Mitigation might be to avoid collection system altogher and focus on treatment of full W15 flow in pit.
	E	Collection of cleaner runoff is inadequate, leading to mixing with water requiring treatment and increased treatment costs	12	Conseq. Costs	Minor	Likely	Moderate	
	A	Seismic or extreme flood event larger than design leads to WSP Dam failure (assumes reduced height), resulting in surge of water and solids into Minto Creek		Env. Imp.	Moderate	Very Unlikely	Low	
	B	WSP Dam (assumes reduced height) maintenance requirements not met, resulting in failure and surge of water and solids into Minto Creek		Env. Imp.	Moderate	Very Unlikely	Low	assumes design with maintenance requirements
	A	Flow rates exceed WTP/surge capacity, resulting in unnacceptable water quality downstream	11	Env. Imp.	Moderate	Possible	Moderately High	Mitigation : increase surge capacity and/or operate surge volumes better - depending on why surge capacity was overwhelmed
	B			Conseq. Costs	Major	Unlikely	Moderately High	assume worst case - plant/surge exceeded because not sufficient
	C	Contaminant loading exceeds treatment capacity, resulting in unnacceptable water quality	11	Env. Imp.	Moderate	Unlikely	Moderate	Function of geochemical source term identification
	D	downstream		Conseq. Costs	Major	Unlikely	Moderately High	
	E	Treatment technology ineffective for contaminants of concern, resulting in unnacceptable	11	Env. Imp.	Moderate	Very Unlikely	Low	
	F	water quality downstream		Conseq. Costs	Major	Very Unlikely	Moderate	
	G	Inadequate capacity for storage of byproducts, leads to costs for removal off site		Conseq. Costs	Moderate	Very Unlikely	Low	
	A	Extreme event leads to failure of conveyance structure upgradient of DSTSF, flow onto DSTSF leading to cover damage and tailings mobilization across top of DSTSF leading to unacceptable water quality conditions downstream	7W, 10, 12, 15	Env. Imp.	Major	Possible	High	upslope key of cover? Pitwall failure should be considered elsewhere in planning. Reducing dependance on manmade structures is desirable-i.e. wingwalls at pit outlets
	B	Leakage from conveyance structure upgradient of DSTSF increases flow subsurface and contaminant loading from tailings leading to unacceptable downstream water quality	7W, 12, 15	Env. Imp.	Moderate	Unlikely	Moderate	these should be designed to reduce leakage/seepage to DSTSF
	2	**Chemical Stability**						
	A	Source water quality (source term) worse than expected and causes unacceptable water quality conditions **downstream of site**	3,7,10	Env. Imp.	Critical	Possible	High	assuming no AMP in place
	B	Source water quality (source term) worse than expected and causes unacceptable water quality conditions **downstream of site**		Env. Imp.	Moderate	Possible	Moderately High	assumed AMP in place so Severity moderate, critical to understand chemistry as fully as possible - reflected in likelihood designation, concern about reliance on AMP
	C			Conseq. Costs	Major	Possible	High	
	D			Spec. Cons.	Moderate	Possible	Moderately High	
	E	Source water quality (source term) worse than expected and causes unacceptable water quality conditions **on site**		Env. Imp.	Moderate	Unlikely	Moderate	

	F		2,3,7,10	Legal Obl.	Major	Possible	High	This could apply to the preceding mode if there were W2 standards in place
	G			Conseq. Costs	Moderate	Possible	Moderately High	
	A	Covers don't perform as designed re: infiltration resulting in unacceptable water quality conditions **downstream of site**	10	Env. Imp.	Moderate	Unlikely	Moderate	Assuming AMP and monitoring will respond to any defects
	B			Conseq. Costs	Major	Unlikely	Moderately High	
	B	Erosion leads to increased infiltration and unacceptable downstream WQ effects	10, 12,15	Env. Imp.	Moderate	Unlikely	Moderate	Risks different for DSTSF than for other facilities, potential effects of erosion still need to be considered in design, maintenance costing, etc.
	A	Bioreactors don't perform as designed - overwhelmed, freeze, resulting in unacceptable water quality conditions downstream	11	Env. Imp.	Moderate	Almost Certain	High	Assumption: effective AMP in place Notes: cryo-concentration in seeps, ice cleaner, residual seeps higher concentration Leslie: make sure that any supporting work here has data - not just stories that they work -i.e. Andre Sobolewski's work at G900 didn't work, but MPERG report still says it does.
	A	Wetlands don't perform as designed - overwhelmed, freeze, resulting in unacceptable water quality conditions downstream	11	Env. Imp.	Moderate	Almost Certain	High	
	B	High flow blow out wetland, causing damage and maintenance requirements, assuming high flows designed to bypass	11, 12	Env. Imp.	Moderate	Unlikely	Moderate	Assumption: peak flows not treated by wetlands - need to understand the implications of this during freshet and also during peak flow events. Wetlands are not designed nor capable of treating peak flows, so this is a significant red-flag for planning - needs very careful consideration.
	C			Conseq. Costs	Moderate	Unlikely	Moderate	Need to understand the implications of this during freshet and also during peak flow events.
	A	Pit Lake Treatment (Non-Flow through Pit)						
	i	Non-flow through Area 2 Pit treatment compromised because of diversion ditch failure, resulting in flow through condition		Env. Imp.	Minor	Possible	Moderate	Assuming pit water quality has moderate initial contamination level - make sure this is covered in AMP. What if WQ in Area 2 pit were higher than anticipated?
	ii	Non-flow through pit treatment does not perform		Env. Imp.	Minor	Possible	Moderate	
	iii	Pit Wall Failure in Area 2 results in wave of water released from pit causing damage to downstream facilities and tailings mobilization from bottom of Area 2 pit		Conseq. Costs	Major	Unlikely	Moderately High	Resolution would be difficult - would mean appropriate sizing of the spillway, locating of facilities downgradient
	B	Pit Lake Treatment (Flow through Pit)						
		Flow through Pit treatment does not perform as expected		Env. Imp.	Minor	Unlikely	Low	Assume treatment expectations consider flow through condition and limitations
		Flow through Pit source term underestimated, resulting in higher than expected loading from pit and unacceptable water quality results downstream	2C	Env. Imp.	Moderate	Unlikely	Moderate	
		Pit Wall Failure in Area 2 results in wave of water released from pit causing damage to downstream facilities and tailings mobilization from bottom of Area 2 pit	2P	Conseq. Costs	Major	Very Unlikely	Moderate	In the flow through pit condition, the downstream channels and facilities would be designed for hitgher flows, so likelihood lower than in the non-flow through condition.
	3	**Physical Stability**						
	A	Geotechnical failure of any waste facility (slope stability) resulting in waste material exposure to water leading to unacceptable water quality conditions downstream of site	2P, 3P, 4P, 6P, 7P, 8P, 11	Env. Imp.	Moderate	Possible	Moderately High	this is a result of there being permafrost considerations under some facilities, and uncertainty associated. Could result from differential settlement of pockets of more moist materials
	B			Conseq. Costs	Moderate	Possible	Moderately High	
	C	Geotechnical failure of any waste facility (differential settlement) resulting in rupture of cover and waste material exposure to water leading to unacceptable water quality conditions downstream of site	2P, 3P, 4P, 6P, 7P, 8P, 10, 11	Env. Imp.	Moderate	Likely	Moderately High	Could result from differential settlement of pockets of high moisture materials. Could be moderated by waste mgmt practices limiting wet waste in waste dumps
	D	Geotechnical failure of any waste facility (slope stability) resulting in debris dam, breaching, mobilizing materials and pulse of water into Main Pit, and sediments/tailings leaving pit, leading to unacceptable water quality conditions downstream of site	2P, 3P, 4P, 6P, 7P, 8P, 10, 11, 12	Env. Imp.	Minor	Possible	Moderate	
	4	**Administrative**						
	A	Failure to implement AMP, resulting in unacceptable water quality conditions downstream	14	Env. Imp.	Critical	Possible	High	
	B	Failure to design an appropriate AMP, resulting in unacceptable water quality conditions downstream	14	Env. Imp.	Major	Unlikely	Moderately High	Important to recognize that AMP is more than just monitoring - but careful identification of potential issues, thresholds and appropriate responses. AMP not just an add-on. Needs to be critical component of closure plan at same detail as rest of plan
	C	Departure from design of engineered structures, resulting in unacceptable water quality conditions downstream	14	Env. Imp.	Major	Possible	High	
	A	General failure to maintain site requirements as required - passive treatment, cover maintenance, etc.	14	Env. Imp.	Major	Possible	High	

	2. OPEN PITS	Consequence		Likelihood	Risk Rating	NOTES (yellow highlighted notes are conditional risk ratings)
		Type	Severity	Probability	Descriptive	
1	**Area 2 Pit**					
W	**Water Management**					
1	Precipitation higher than design asssumption (1:200 yr 24 hr) resulting in damage to outlet structure	Conseq. Costs	Minor	Unlikely	Low	Assumes complete replacement of spillway; review whether climate change (potential for inceased precip) has been allowed for
2	Precipitation higher than design asssumption (1:200 yr 24 hr) resulting in erosion of Ditch 400 channel and damage to toe of Main Dam, leading to breach of dam and release of tailings to lower Minto Creek	Env. Imp.	Major	Very Unlikely	Moderate	
3	Precipitation higher than design asssumption (1:200 yr 24 hr) resulting in erosion of Ditch 400 channel and damage to toe of Main Dam, leading to breach of dam and release of tailings to lower Minto Creek	Trad. Use	Major	Very Unlikely	Moderate	
4	Precipitation higher than design asssumption (1:200 yr 24 hr) resulting in erosion of Ditch 400 channel and damage to toe of Main Dam, leading to breach of dam and release of tailings to lower Minto Creek	Reg. & Legal	Major	Very Unlikely	Moderate	
5	Precipitation higher than design asssumption (1:200 yr 24 hr) resulting in erosion of Ditch 400 channel and damage to toe of Main Dam, leading to breach of dam and release of tailings to lower Minto Creek	Conseq. Costs	Critical	Very Unlikely	Moderately High	
6	Precipitation higher than design asssumption (1:200 yr 24 hr) resulting in erosion of Ditch 400 channel and damage to toe of Main Dam, leading to breach of dam and release of tailings to lower Minto Creek	Comm/Media/Rep	Critical	Very Unlikely	Moderately High	
7	Precipitation higher than design asssumption (1:200 yr 24 hr) resulting in erosion of Ditch 400 channel and damage to toe of Main Dam, leading to breach of dam and release of tailings to lower Minto Creek	Human H&S	Moderate	Very Unlikely	Low	
P	**Physical Stability**					
1	Pit wall failure in Area 2 results in wave of water released from pit causing damage to downstream facilities (ditches, passive treatment system, covers)	Env. Imp.	Moderate	Unlikely	Moderate	-Outlet of pit spillway is ~6m deep, water depth in pit ~35m. -Large degree of uncertainty regarding the likelyhood of failure. -Could be mitigated by sharing of wall stability information
2	Pit wall failure in Area 2 results in wave of water released from pit causing damage to downstream facilities (ditches, passive treatment system, covers)	Conseq. Costs	Moderate	Unlikely	Moderate	Same as above
3	Pit wall failure in Area 2 results in wave of water released from pit and causes a fatality.	Human H&S	Critical	Very Unlikely	Moderately High	Same as above; -Ranking preliminary; Failure has not been evaluated. -If a lower likelyhood option was available, it would have been selected.
C	**Chemical Stability**					
1	Pit water quality at a level that causes unacceptable water quality conditions for water fowl / wildlife	Env. Imp.	Moderate	Unlikely	Moderate	Scenario needs to be evaluated; water quality that could affect water fowl needs to be researched and shared.
2	Pit water quality at a level that causes unacceptable water quality conditions for water fowl / wildlife	Trad. Use	Moderate	Unlikely	Moderate	Same as above;
3	Pit water quality at a level that leads to problematic exceedances of downstream water quality objectives	Env. Imp.	Moderate	Possible	Moderately High	Assumes AMP in place, funded (results in a short term impact)
4	Pit water quality at a level that leads to problematic exceedances of site water quality discharge standards	Reg. & Legal	Moderate	Possible	Moderately High	
5	Pit limnology leads to problematic exceedances of site water quality discharge standards	Env. Imp.	Moderate	Unlikely	Moderate	Scenario needs to be evaluated;
6	Discharge water quality objectives are not met when pit first discharges	Env. Imp.	Moderate	Very Unlikely	Low	Pit water quality assumed to be carefully monitored during the transition stage; (pit takes ~ 3years to fill); Assumes treatment occurs if requried.

7	Discharge water quality objectives are not met when pit first discharges requiring treatment to meet discharge objectives by time of first discharge	Conseq. Costs	Moderate	Possible	Moderately High	Pit water quality assumed to be carefully monitored during the transition stage; (pit takes ~ 3years to fill)
2	**Minto North Pit**					
W	**Water Management**					
1	Positive water balance for pit leads to overtopping and erosion at spill point and sediment release downstream, causing unacceptable sedimentation in Mcginty Creek	Env. Imp.	Minor	Unlikely	Low	This is a good item for inclusion in the AMP. Observations during operations and Post-closure 1 period will be useful in better understanding future fate of MN Pit water balance
2	Positive water balance for pit leads to development of pit lake and results in negative perception leading to negative impacts on traditional land use	Trad. Use	Minor	Possible	Moderate	Could be mitigated through education and information sharing. Discussion by group noted that mitigation of perception is complicated by variability in perceptions among individuals.
P	**Physical Stability**					
1	Massive rapid pit wall failure into pit full of water results in wave of water spilling over rim of pit causing erosion and downstream sedimentation and riparian damage	Env. Imp.	Minor	Very Unlikely	Low	
C	**Chemical Stability**					
1	Pit water quality at a level that causes unacceptable water quality conditions for water fowl / wildlife	Env. Imp.	Moderate	Very Unlikely	Low	Scenario needs to be evaluated; water quality that could affect water fowl needs to be researched and shared. Scenario requires pit to contain water, which is uncertain
2	Pit water quality at a level that leads to problematic exceedances of downstream water quality objectives	Env. Imp.	Moderate	Possible	Moderately High	Assumes AMP does not exist or is not implemented
3	Pit water quality at a level that causes unacceptable water quality conditions for water fowl / wildlife and results in some mitigatable impact to tradtional land use	Trad. Use	Moderate	Very Unlikely	Low	Same as above.
3	**Ridgetop North Pit**					
W	**Water Management**					
1	Erosion of downstream slope due to runoff from covered tailings leads to need for repairs and/or establishment of conveyance structure	Conseq. Costs	Minor	Possible	Moderate	Could mitigated through AMP or addressed in closure plan
2	Settlement of tailings leads to ponding on surface of covered tailings resulting in increased infiltration and leading unacceptable downstream water quality	Env. Imp.	Moderate	Very Unlikely	Low	
P	**Physical Stability**					
1	High wall slope failure leads to cover damage, exposure of tailings and need for cover repairs	Conseq. Costs	Minor	Very Unlikely	Low	
2	Ponding of water within RNPTMF against E/NE overburden wall causes slope instability in overburden	Conseq. Costs	Moderate	Unlikely	Moderate	Could be mitigated by including regrading to eleminate ponding in AMP; could undertake slope stability analysis to evaluate risk of this failure mode
3	Thawing of entrained ice leads to settlement in the pit and ponding of water within RNPTMF against E/NE overburden wall causes slope instability in overburden	Conseq. Costs	Moderate	Unlikely	Moderate	Could be mitigated through appropriate tailings deposition plan
C	**Chemical Stability**					
1	No facility specific scenarios idenfied - covered under '10 Source Control'				#N/A	

	3. WASTE ROCK DUMPS	Consequence		Likelihood	Risk Rating	NOTES
		Type	Severity	Probability	Descriptive	
1	**Main Waste Dump**					
W	Water Management					
1	Precipitation higher than expected resulting in failure of water conveyance structures because structures are underdesigned				#N/A	Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA
2	Undiverted runoff upstream of waste mgmt facilities leads to runon water, extra infiltration, leading to ongoing maintenance costs				#N/A	Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA
3	Run-on water from upgradient catchment of MWD increases flow subsurface and contaminant loading from waste rock leading to unacceptable downstream water quality	Env. Imp.	Moderate	Very Unlikely	Low	MWD has limited upgradient catchment area and limited resulting run-on
4	Ponding of water on surface of MWD leads to excessive infiltration, increases flow subsurface and contaminant loading from waste rock leading to unacceptable downstream water quality	Env. Imp.	Moderate	Unlikely	Moderate	Requires failure of AMP to realize the scenario. Can be mitigated by shaping surface to limit ponding; maintenance may be required to restore contouring (if differential settlement occurs) to prevent ponding
P	Physical Stability					
1	Instability results in waste material exposure to water leading to unacceptable downstream water quality	Env. Imp.	Moderate	Very Unlikely	Low	Likelihood supported by stability evaluations in MWDE design report
2	Erosion on steeper portion of MWD leads to loss of cover and results in need for repairs	Conseq. Costs	Minor	Likely	Moderate	Can be mitigated through design and/or O&M plan
3	Die-back of cover vegetation after successful establishment and acceptance leads to erosion, and need for repair	Conseq. Costs	Minor	Possible	Moderate	MWD will have been covered/ planted for several years; appropriate selection of veg species would reduce chance of wholesale die-back. Can be mitigated through design and/or O&M plan
4	Root throw results in increased infiltration over the long term and leads to need for repair	Conseq. Costs	Minor	Unlikely	Low	Can be mitigated through design (including appropriate selection of veg species) and/or O&M plan
2	**Southwest Waste Dump**					
W	Water Management					
1	SWD toe seepage collection systems inadequate, leading to unacceptable WQ downstream				#N/A	Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA
2	Precipitation higher than expected resulting in failure of water conveyance structures because structures are underdesigned				#N/A	Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA
3	Undiverted runoff upstream of waste mgmt facilities leads to runon water, extra infiltration, leading to ongoing maintenance costs				#N/A	Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA
4	Run-on water from upgradient catchment of SWD increases flow subsurface and contaminant loading from waste rock leading to unacceptable downstream water quality	Env. Imp.	Moderate	Very Unlikely	Low	SWD has limited upgradient catchment area and limited resulting run-on
5	Ponding of water on surface of SWD leads to excessive infiltration, increases flow subsurface and contaminant loading from waste rock leading to unacceptable downstream water quality	Env. Imp.	Moderate	Unlikely	Moderate	Requires failure of AMP to realize the scenario. Can be mitigated by shaping surface to limit ponding; maintenance may be required to restore contouring (if differential settlement occurs) to prevent ponding. There is a BGM cover on HGW, so ponding+increased infiltration is less likely for HGW
6	Existing pond north of MGW/ south of IROD remains in post-closure and causes community concern	Comm/Media/Rep	Very Low	Almost Certain	Moderate	Will be revisited in detailed design and could be mitigated through education and information sharing
P	Physical Stability					
1	Instability results in waste material exposure to water leading to unacceptable downstream water quality	Env. Imp.	Moderate	Very Unlikely	Low	Likelihood supported by stability evaluations in SWD design report??? Foundation includes permafrost overburden, but the design considered the existing foundation conditions

2	Erosion on steeper portions of SWD leads to loss of cover and results in need for repairs	Conseq. Costs	Minor	Possible	Moderate	Can be mitigated through design and/or O&M plan
3	Die-back of cover vegetation after successful establishment and acceptance leads to erosion, and need for repair	Conseq. Costs	Minor	Possible	Moderate	SWD will have been covered/ planted for several years; appropriate selection of veg species would reduce chance of wholesale die-back. Can be mitigated through design and/or O&M plan
4	Root throw results in increased infiltration over the long term and leads to need for repair	Conseq. Costs	Minor	Unlikely	Low	Can be mitigated through design (including appropriate selection of veg species) and/or O&M plan
5	Root throw results in damage to engineered cover (BGM) over HGW leads to increased infiltration over the long term and leads to need for repair	Conseq. Costs	Minor	Possible	Moderate	Can be mitigated through design (including appropriate selection of veg species) and/or O&M plan
3	**Ridgetop Waste Dump**					
W	Water Management					
1	Precipitation higher than expected resulting in failure of water conveyance structures because structures are underdesigned				#N/A	Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA
2	Undiverted runoff upstream of waste mgmt facilities leads to runon water, extra infiltration, leading to ongoing maintenance costs				#N/A	Not rated due to absence of upgradient catchment by design
3	Run-on water from upgradient catchment of RWD increases flow subsurface and contaminant loading from waste rock leading to unacceptable downstream water quality				#N/A	Not rated due to absence of upgradient catchment by design
4	Ponding of water on surface of RWD leads to excessive infiltration, increases flow subsurface and contaminant loading from waste rock leading to unacceptable downstream water quality	Env. Imp.	Moderate	Very Unlikely	Low	Requires failure of AMP to realize the scenario. Can be mitigated by shaping surface to limit ponding; maintenance may be required to restore contouring (if differential settlement occurs) to prevent ponding.
P	Physical Stability					
1	Instability results in waste material exposure to water leading to unacceptable downstream water quality	Env. Imp.	Moderate	Very Unlikely	Low	Likelihood supported by stability evaluations in RWD design report. Foundation does not include permafrost overburden
2	Erosion on steeper portions of RWD leads to loss of cover and results in need for repairs	Conseq. Costs	Minor	Possible	Moderate	Can be mitigated through design and/or O&M plan
3	Die-back of cover vegetation after successful establishment and acceptance leads to erosion, and need for repair	Conseq. Costs	Minor	Possible	Moderate	RWD will have been covered/ planted for several years; appropriate selection of veg species would reduce chance of wholesale die-back. Can be mitigated through design and/or O&M plan
4	Root throw results in increased infiltration over the long term and leads to need for repair	Conseq. Costs	Minor	Unlikely	Low	Can be mitigated through design (including appropriate selection of veg species) and/or O&M plan
5	Ridgetop Waste Dump name creates perception that there will be major viewshed impacts	Comm/Media/Rep	Minor	Possible	Moderate	Could be mitigated through education and information sharing
4	**Reclamation Overburden Dump, Ridgetop South and Area 118 Backfill Dumps**					
W	Water Management					
1	Precipitation higher than expected resulting in failure of water conveyance structures because structures are underdesigned				#N/A	Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA
2	Undiverted runoff upstream of waste mgmt facilities leads to runon water, extra infiltration, leading to ongoing maintenance costs				#N/A	Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA
3	Run-on water from upgradient catchments increases flow subsurface and contaminant loading from waste rock leading to unacceptable downstream water quality	Env. Imp.	Moderate	Very Unlikely	Low	Dump contents are overburden
4	Ponding of water on surface of overburden dumps leads to excessive infiltration, increases flow subsurface and contaminant loading from overburden leading to unacceptable downstream water quality	Env. Imp.	Moderate	Very Unlikely	Low	Designs of A118 and RS BD are mounded to shed water.

5	RS BD is not constructed to design limits, resulting in development of a pit lake within RS BD pit leading to spill of pit water and erosion of downgradient slope leading to need for development of channel	Conseq. Costs	Minor	Very Unlikely	Low	Likely will not form lake based on lack of water encountered in A118 pit. Could be mitigated by filling with waste rock during mining or by filling later with ob or waste rock
P	Physical Stability					
1	Erosion on steeper portions of dumps leads to sedimentation in conveyance channels and need for maintenance				#N/A	Scenario topic covered in #12 in 2014 FMEA
2	Die-back of vegetation after successful establishment and acceptance leads to erosion, and need for repair	Conseq. Costs	Minor	Unlikely	Low	No requirement to maintain cover integrity due to dump material (ob)
3	Storage of overburden in pits creates perception that valuable reclamation materials are being wasted	Comm/Media/Rep	Minor	Almost Certain	Moderately High	Could be mitigated through education and information sharing

	6. MAIN DAM	Consequence		Likelihood	Risk Rating	NOTES
		Type	Severity	Probability	Descriptive	
1	Blockage or settlement of Ditch 300 leads to water in Main Pit and the Main dam has settled, leading to opertopping of the Main Dam resulting in breach, release of tailings and water to lower Minto Creek	Env. Imp.	Major	Possible	High	Assumes Main Dam has settled and/or blockage of the spillway -Could be mitigated by adding material to dam to account for settlement and/or reevaluation of spillway
2	Blockage or settlement of Ditch 300 leads to water in Main Pit and the Main dam has settled, leading to opertopping of the Main Dam resulting in breach, release of tailings and water to lower Minto Creek	Comm/Media/Rep	Critical	Possible	High	Assumes Main Dam has settled and/or blockage of the spillway -Could be mitigated by adding material to dam to account for settlement and/or reevaluation of spillway -Any breach of tailings would be considered a critical severity to the local community
3	Thawing of entrained ice leads to settlement in the pit and ponding of water against the dam leading to excessive seepage and piping resulting in dam failure and release of tailings that remain within the mine site and water is released to lower Minto Creek.	Env. Imp.	Moderate	Unlikely	Moderate	Rockfill dam not likely to pipe.
4	Thawing of entrained ice leads to settlement in the pit and ponding of water against the dam leading to excessive seepage and piping resulting in dam failure and release of tailings that remain within the mine site and water is released to lower Minto Creek.	Conseq. Costs	Moderate	Unlikely	Moderate	Rockfill dam not likely to pipe.
5	Rapid/Massive failure of the Pit high wall resulting in material entering the pit and displacing tailings that remain within the pit.	Conseq. Costs	Minor	Unlikely	Low	
6	Shallow toe failure in the thawed ground leading to slumping and damage to the dam resulting in need for repair.	Conseq. Costs	Minor	Possible	Moderate	
7	Continued movement along the shear zone at the DSTSF leads to movement of the shear zone near the dam leading to cracking of the core and failure of the liner leading to breach release of tailings and water to lower Minto Creek	Env. Imp.	Major	Very Unlikely	Moderate	
8	Earthquake leads to liquefaction of foundation soils resulting in downstream slope failure resulting in a breach and release of tailings and water to lower Minto Creek	Env. Imp.			#N/A	Not rated. High level of uncertainty - needs to be investigated
9	Failure of the north wall of the Area 2 Pit leads to failure of the south abutment and progressive failure of the dam resulting in release of tailings and water to lower Minto Creek	Env. Imp.			#N/A	Not rated. High level of uncertainty - needs to be investigated

	7. DRY STACK TAILINGS STORAGE FACILITY	Consequence		Likelihood	Risk Rating	NOTES
		Type	Severity	Probability	Descriptive	
1	Long-term permafrost degradation underlying the DSTSF leads differential settlement and ponding of water resulting in increased infiltration leading to unacceptable downstream water quality	Env. Imp.	Minor	Unlikely	Low	
2	Long-term permafrost degradation underlying the DSTSF leads differential settlement and ponding of water resulting in need for repair	Conseq. Costs	Minor	Possible	Moderate	
3	Long-term permafrost degradation underlying the DSTSF leads differential settlement and ponding of water resulting in erosion of cover materials and sediment loading to passive treatment system	Env. Imp.	Minor	Possible	Moderate	Maintenance issue
4	Re-initiation of shear zone due to thawing of permafrost from the bottom up due to high excess pore pressure and movement in the cross- valley direction, leading to cracking of the cover, increased infiltration and impacts to downstream water quality.	Env. Imp.	Minor	Very Unlikely	Low	Mitigated by construction of MVFE S2
5	Deep-seated downvalley slope failure of the MVFE (Section F) leads to instability of the DSTSF and cracking of the cover, failure of conveyance ditches and need for additional stabilization measures.	Conseq. Costs	Major	Very Unlikely	Moderate	MVFE S2 design report calculates a minimum FOS of 2.3. MVFE S2 will be constructed years before closure resulting in years of performance monitoring.
6	MVFE S2 does not extend far enough downvalley to prevent cross-valley movement at the eastern extent of the current movement affecting the DSTSF.	Env. Imp.			#N/A	Not rated. To be investigated by the designer.
7	Toe failure of the MVFE (Section F) leads to instability and cracking of the cover, failure of conveyance ditches and need for additional stabilization measures.	Env. Imp.			#N/A	Not rated. To be investigated by the designer.
8	Landslide dam forms in footprint of Water Storage Pond as a result of thawing of permafrost overburden in S valley wall due to presence of Water Storage Pond leads to impounding of water and subsequent rapid breach and sediment loading downstream	Env. Imp.	Minor	Possible	Moderate	Consider including in post-closure monitoring
9	Movement (lateral movement or differential settlement) reduces or blocks flow from the finger drains underlying the DSTSF results in increased pore pressure within tailings mass, raising of water table within tailings mass and ultimately increased daylighting of groundwater upgradient of DSTSF	Env. Imp.	Minor	Very Unlikely	Low	

	10. Source Characterization and Control	Consequence		Likelihood	Risk Rating	NOTES
		Type	Severity	Probability	Descriptive	
1	Source water quality (source term) worse than predicted resulting in problematic exceedances of downstream water quality objectives	Env. Imp.	Major	Unlikely	Moderately High	Severity requires that no AMP response occurs
2	Source water quality (source term) worse than predicted resulting in problematic exceedances of downstream water quality objectives	Conseq. Costs	Critical	Unlikely	Moderately High	Assumes either active treatment or implementation of high quality covers
3	Source water quality (source term) worse than predicted resulting in problematic exceedances of downstream water quality objectives	Comm/Media/Rep	Critical	Unlikely	Moderately High	

	11. WATER TREATMENT	**Consequence**		**Likelihood**	**Risk Rating**	**NOTES**
		Type	**Severity**	**Probability**	**Descriptive**	
1	Passive treatment performance is not sufficient to meet downstream closure water quality objectives				#N/A	After considerable discussion, workshop concluded that the current state of information on the topic of passive treatment does not support assigning likelihood or severity ratings to scenarios around failure of passive treatment.

	12. WATER CONVEYANCE	Consequence		Likelihood	Risk Rating	NOTES
		Type	Severity	Probability	Descriptive	
T	**Tertiary Channels**					
1	Differential settlement leading to excessively concentrated flows in channels leading to erosion and filling of the energy dissipator structures	Conseq. Costs	Minor	Likely	Moderate	Assuming occurs 8 years after closure Mitigatable by performing routine O&M, encorporating rock into soil.
2	Differential settlement leading to excessively concentrated flows in channels leading to waste rock exposure and infiltration leads to unacceptable water quality downstream	Env. Imp.	Minor	Possible	Moderate	Mitigatable by performing routine O&M, encorporating rock into soil.
3	Differential settlement leading to excessively concentrated flows in channels leading to waste rock exposure and infiltration leading to repair requirements	Conseq. Costs	Minor	Likely	Moderate	Mitigatable by performing routine O&M, encorporating rock into soil.
4	Inadequate design of tertiary network results in need for repairs	Conseq. Costs	Minor	Possible	Moderate	Mitigatable by performing routine O&M, encorporating rock into soil. -Further clarification of the design event to be completed.
5	Vegetation growth is less than expected leading to concentrated flows resulting in erosion and filling of the energy dissipator structures	Conseq. Costs	Minor	Possible	Moderate	Mitigatable by performing routine O&M, encorporating rock into soil.
S	**Secondary**					
1	Excessive concentrated flows lead to erosion and gulley formation and mass wasting leading to unacceptable sediment load downstream	Env. Imp.	Minor	Possible	Moderate	
2	Excessive concentrated flows lead to erosion and gulley formation leading to waste rock exposure and infiltration leads to unacceptable water quality downstream	Env. Imp.	Minor	Possible	Moderate	
3	Excessive concentrated flows lead to erosion and gulley formation leading to repair requirements	Conseq. Costs	Minor	Possible	Moderate	
4	Inadequate design of secondary network results in need for repairs	Conseq. Costs	Minor	Possible	Moderate	-Further clarification of the design event to be completed.
P	**Primary**					
1	Instability of the MPD leads to a breach of Ditch 300 resulting in erosion of MPD waste rock cover and exposure of waste rock and infiltration leading to unacceptable water quality downstream	Env. Imp.	Minor	Unlikely	Low	
2	Instability of the MPD leads to a breach of Ditch 300 resulting in erosion of MPD waste rock cover and exposure of waste rock and infiltration leading to repair requirements	Conseq. Costs	Minor	Possible	Moderate	Assumes repair and not a wholescale repair in design
3	Inadequate design of primary network results in need for repairs	Conseq. Costs	Moderate	Unlikely	Moderate	
4	Flows exceed channel capacity resulting in failure of water conveyance structures resulting in a need for repairs	Conseq. Costs	Moderate	Unlikely	Moderate	-Further clarification of the design event to be completed.
5	SWD toe seepage collection systems inadequate, leading to unacceptable WQ downstream	Env. Imp.	Moderate	Almost Certain	High	Not part of RCP v5.1rated, subject of reclamation research; can be mitigated by inclusion
6	Extreme event leads to failure of TDD conveyance structure upgradient of DSTSF, flow onto DSTSF leading to cover damage and tailings mobilization across top of DSTSF leading to unacceptable water quality conditions downstream	Env. Imp.	Moderate	Very Unlikely	Low	Requires erosion of >3m of cover
7	Leakage from TDD conveyance structure upgradient of DSTSF increases flow subsurface and contaminant loading from tailings leading to unacceptable downstream water quality	Env. Imp.	Moderate	Unlikely	Moderate	
8	Winter ice development in primary channels results in inadequate capacity to pass freshet flows resulting in need for repairs.	Conseq. Costs	Moderate	Possible	Moderately High	May be a recuring event. Risk is lower where foundation is mine fill.

PS	**Physical Stability**					
1	Thaw degradation leading to differential settlment of spillway, distruption of the armour layer, and scour of the spillway resulting in the need for repairs	Conseq. Costs	Moderate	Possible	Moderately High	May be a recuring event. Configuation of spillway at closure needs to be considered. Consider emergency spillway during operations and different long term spillway. (consider in rock on north side). Likelyhood could be mitigated with routine maintenance.
2	Thaw degradation leads to retrogressive failure at outlet of main pit spillway resulting in the need for repairs.	Conseq. Costs	Moderate	Possible	Moderately High	same as above
3	Thaw degradation leading to differential settlement at the outlet of the Area 2 Pit resulting in the need for repairs.	Conseq. Costs	Minor	Likely	Moderate	
4	Thaw degradation leading to differential settlment in Ditch 400 downstream of the outlet leading to distruption of the armour layer, and scour of the channel resulting in the need for repairs	Conseq. Costs	Minor	Possible	Moderate	
5	Thaw degradation leading to differential settlement in Ditch 400 downstream of the outlet leading to ponding, overtopping, and erosion resulting in the need for repairs.	Conseq. Costs	Minor	Unlikely	Low	To be checked by designers
6	Thaw degradation leading to differential settlement in Ditch 400 downstream of the outlet leading to ponding, overtopping and erosion resulting in sediment loading into passive treatment system leading to impacts to water quality.	Env. Imp.	Moderate	Very Unlikely	Low	
7	Geotechnical failure of any waste facility (slope stability) resulting in debris dam, breaching, mobilizing materials and pulse of water into Main Pit, and sediments/tailings leaving pit, leading to unacceptable water quality conditions downstream of site	Env. Imp.	Very Low	Very Unlikely	Low	Revisit on Aug 28
C	**Chemical Stability**					
1	Increased infiltration through unlined Ditch 200 and Ditch 300 leads to unacceptable water quality conditions downstream of site	Env. Imp.	Moderate	Very Unlikely	Low	

	14. ADMINISTRATIVE	Consequence		Likelihood	Risk Rating	NOTES
		Type	Severity	Probability	Descriptive	
1	General failure to conduct preventative mainenance and corrective actions leading to system failures (passive treatment, covers, etc.) resulting in impacts on ecosystem components.	Env. Imp.	Major	Likely	High	Impacts on Minto Creek. Reason for this rating is a lack of O&M plan beyond year 13. Some uncertainty resulting from effects of permafrost degradation and climate change.
2	General failure to conduct preventative mainenance and corrective actions leading to system failures (passive treatment, covers, etc.) resulting in negative traditional use	Trad. Use	Critical	Likely	Very High	-Assumes the existing traditional land use impacted due to perception; cease in trapping, hunting, berry gathering activity in area. -Assumes intended post-mining land use same as pre-mining land use
3	General failure to conduct preventative mainenance and corrective actions leading to system failures (passive treatment, covers, etc.) resulting in regulatory/legal action.	Reg. & Legal	Critical	Possible	High	-Assumes company remains the responsible company.
4	General failure to conduct preventative mainenance and corrective actions leading to system failures (passive treatment, covers, etc.) resulting in ….	Conseq. Costs			#N/A	Not rated. In the event that this occurs there is likely to be a large public liability
5	General failure to conduct preventative mainenance and corrective actions leading to system failures (passive treatment, covers, etc.) resulting in community/media/reputation impacts	Comm/Media/Rep	Critical	Likely	Very High	media and reputation aspects of the concequence severity description were disregarded in rating.
6	General failure to conduct preventative mainenance and corrective actions leading to system failures (passive treatment, covers, etc.) resulting in H&S impacts	Human H&S	Minor	Unlikely	Low	
7	Departure from design of engineered structures, resulting in unacceptable water quality conditions downstream	Env. Imp.	Moderate	Possible	Moderately High	Can be mitigated with appropriate QA/QC monitoring
8	Departure from design of engineered structures, resulting in need for upgrades/ repairs/ redesign	Conseq. Costs	Major	Possible	High	Can be mitigated with appropriate QA/QC monitoring
9	Bankruptcy/ dissolution of the company and inadequate financial security leads to requirement for public government to fund and conduct preventative mainenance and corrective actions to avoid system failures and impacts on ecosystem components.	Conseq. Costs	Major	Possible	High	Likelihood rating was selected to reflect that this scenario has happened elsewhere
10	Revegetation does not meet closure objectives relating to end land use	Trad. Use	Moderate	Possible	Moderately High	Can be mitigated through appropriate selection of end land use, determination of end land use goals, appropriate development of closure objectives to support those goals, and appropriate selection of veg species and revegetation methods
11	Failure of institutional controls resulting in land use that causes unanticipated negative exposure of humans or wildilife	Human H&S	Critical	Very Unlikely	Moderately High	Rated assuming failure results in a human fatality. Likelihood rating considers remoteness of site.
12	Failure to adequately meet reporting requirements results in noncompliance	Reg. & Legal	Minor	Possible	Moderate	

APPENDIX D

FMEA SUMMARY RESULTS

Summary of Risk Ranking for All Scenarios Considered

Likelihood	Consequence Severity				
	Very Low	Minor	Moderate	Major	Critical
Almost Certain	3.2-W6	3.4-P3	12-P-5		
Likely		12-T-1, 12-T-3, 12-PS-3, 3.1-P2		14-1	14-5, 14-2
Possible		12-T-2, 12-T-4, 12-T-5, 12-S-1, 12-S-2, 12-S-3, 12-S-4,12-P-2, 12-PS-4, 6-6, 7-2, 7-3, 7-8, 2.2-W2, 2.3-W1, 3.1-P3, 3.2-P2, 3.2-P3, 3.2-P5, 3.3-P2, 3.3-P3, 3.3-P5, 14-12	12-P-8, 12-PS-1, 12-PS-2, 2.1-C3, 2.1-C4, 2.1-C7, 2.2-C2, 14-7, 14-10	6-1, 14-8, 14-9	6-2, 14-3
Unlikely		12-P-1, 12-PS-5, 6-5, 7-1, 2.1-W1, 2.2-W1, 3.1-P4, 3.2-P4, 3.3-P4, 3.4-P2, 14-6	12-P-3, 12-P-7, 12-P-4, 6-3, 6-4, 2.1-P1, 2.1-P2, 2.1-C1, 2.1-C2, 2.1-C5, 2.3-P2, 2.3-P3, 3.1-W4, 3.2-W5, 3.3-W4	10-1	10-2, 10-3
Very Unlikely	12-PS-7	12-PS-6, 7-4, 7-9, 2.2-P1, 2.3-P1, 3.4-W5	12-C-1, 12-P-6, 2.1-W7, 2.1-C6, 2.2-C1, 2.2-C3, 2.3-W2, 3.1-W3, 3.1-P1, 3.2-W4, 3.2-P1, 3.3-W4, 3.3-P1, 3.4-W3, 3.4-W4	6-7, 7-5, 2.1-W2, 2.1-W3, 2.1-W4	2.1-W5, 2.1-W6, 2.1-P3, 14-11

Legend:

Descriptors: Site Area – Category – Scenario (e.g. 12-T-1). ** Some descriptors are simplified to Site Area – Scenario (e.g. 14-5)

Site Areas: 2 – Pits, 3 – Waste Rock Dumps, 6 – Main Dam, 7 – Dry Stack Tailings Storage, 10 – Source Control, 11 – Water Treatment, 12 – Water Conveyance, 14 – Administrative

Category: W – Water Management, P - Physical Stability, C - Chemical Stability, A – Administrative

Site Area 12 (Water Conveyance) is unique in its nomenclature in that it used slightly different modifiers, as follows:
P: Primary, S: Secondary, T: Tertiary and PS: Physical Stability

Scenario: Refer to Risk Scenarios in Appendix C for description of the scenarios considered.

MINTO MINE CLOSURE – FAILURE MODES AND EFFECTS ASSESSMENT

2017 WORKSHOP REPORT

MIN-16-03-_008_Minto Mine Closure FMEA 2017 Final Report_RevA_180123

February 1, 2018

Prepared for:

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ALEXCO ENVIRONMENTAL GROUP INC. SIGNATURES

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LIST OF ACRONYMS AND ABBREVIATIONS

ACG	Access Consulting Group
AEG	Alexco Environmental Group Inc.
AES	Atmospheric Environment Service
EMR	Energy Mines and Resources
FMEA	Failure Modes and Effects Analysis
MWD	Main Waste Dump
PCMM	Post Closure Monitoring and Maintenance
RCP	Reclamation and Closure Plan
SAT	Waste Rock that has a NP:AP <3
SFN	Selkirk First Nation
WQ	Water Quality
WUL	Water Use Licence
YESAA	Yukon Environmental and Socio-economic Assessment Act
YESAB	Yukon Environmental and Socio-economic Assessment Board
YG	Yukon Government
YWB	Yukon Water Board

1. INTRODUCTION

Failure Modes and Effects Assessment (FMEA) is a risk evaluation tool that has been used to identify and rank risks associated with the Minto Mine Reclamation and Closure Plan (RCP) since 2013. This report provides a brief background of these exercises and presents a summary of the 2017 FMEA update, which was undertaken to evaluate risks associated with Minto's RCP v2017-01, and to inform planning and mitigative measure selection in RCP v2018-01.

1.1 BACKGROUND

In 2013, in advance of the submission of a Reclamation and Closure Plan (RCP) for the Phase IV mine plan at the Minto Mine, Minto Explorations Ltd. (Minto) conducted a multi-stakeholder Failure Modes and Effects Assessment (FMEA) on a suite of example mine closure scenarios. The workshop was held in Whitehorse in January of 2013, and involved participants representing Minto, Selkirk First Nation (SFN), Yukon Government- Energy Mines and Resources (YG-EMR), and the Yukon Water Board (YWB).

In support of permit amendment applications for the Phase V/VI Expansion mine plan, Minto prepared an updated Reclamation and Closure Plan (RCP v5.1, August 2014). Minto hosted a multi-stakeholder FMEA workshop (in two parts) for the Phase V/VI RCP. The first session was held in Vancouver in August 2014. The two-day workshop in August 2014 included participation by representatives of Minto, SFN, and YG-EMR. Representatives from Norwest Corporation (in its third-party review capacity on geotechnical subjects at Minto Mine, on behalf of Minto and SFN jointly) participated on the second day only. A second, supplementary FMEA workshop was held in Whitehorse in October 2014 with a smaller subset of the original workshop group to address outstanding mine components and closure aspects that had not been addressed in the first workshop. The 2014 FMEA Report was included as an appendix to subsequent RCP versions (see Section 1.2).

1.2 CURRENT PLANNING

Per its Water Use Licence condition, Minto submitted an updated Reclamation and Closure Plan in August 2016 (RCP v2016-01) which incorporated changes identified as per the WUL and QML. It included the advancement of closure designs from conceptual to preliminary, as well as an updated water and load balance model, and addressed the narrowed mine plan scope compared to earlier versions of the RCP. A further updated RCP was submitted in March 2017 (RCP v2017-01, March 2017) which incorporated the final Post Closure Water Quality Objectives and an updated Closure Adaptive Management Plan (AMP).

These RCP versions have been submitted to the YWB as a WUL amendment application, and as such have been subject to an adequacy review. On May 26, 2017, YWB issued their Technical Information Request #2, which included the following related to the FMEA:

> ***IR2-15. Please provide an updated closure FMEA that includes all components of the proposed RCO, including water treatment, and deals wit the "parking lot" issues identified on Page 4 of exhibit 1.2 [the FMEA summary report].***

These parking lot issues were items raised in the FMEA for which either not enough information existed regarding RCP details to be able to effectively evaluate the risk (e.g. passive water treatment plan) or the team could not determine an effective way to resolve a difference of opinion, and a 'note' was made in the parking lot. It was identified that the following parking lot items required further consideration:

- Current closure plan is deficient with respect to showing final reclaimed facilities, toes of re-graded slopes and location of secondary and tertiary water conveyance;
- More information requested regarding the status of the Reclamation Research Plan and the Main Waste Dump revegetation trials;
- Trafficability layer is required over the Ridgetop North Pit tailings backfill whereas costing only allows for 0.5m of overburden;
- Need to advance the discussion/determination of what constitutes "reasonable and practicable" passive treatment, establish protocols and revisit the options evaluation;
- SFN reiterated their concern that the consequence category of "Community/Media/Reputation" is biased because SFN are lumped together with groups having other interests and perspective; and
- The current closure plan does not sufficiently address signage and access control. There is a need to retain institutional controls and maintain signage in perpetuity.

This report provides a description of the 2017 FMEA Workshop proceedings and outcomes, undertaken to meet these YWB information requests, and to inform ongoing planning for the next version of the Minto RCP. It includes the workshop objectives and scope, the approach to the workshop, a description of the workshop activities and the final risk registers developed in the exercise.

2. 2017 FMEA WORKSHOP

In support of the planned RCP revisions and their submission to YWB in the form of RCP v2018-01, and in an effort to supply YWB with the requested information, Minto conducted the first FMEA session in Vancouver on October 18th and 19th, 2017. This was an internal exercise, and is effectively a re-evaluation of the 2014 risk register in the context of any new information. The first (of two) workshops covered site components for which a sufficient level of design currently exists to facilitate this risk assessment. It was focused on site infrastructure (pits, waste rock dumps, dry stack tailings storage facility, water conveyance, covers) and administrative topics. The second workshop was held December 1, 2017 to cover water quality/water treatment topics.

2.1 WORKSHOP OBJECTIVES

The primary objective of the FMEA workshop was to characterize the residual risks that would remain after implementation of the RCP v2017-01, particularly in the context of higher identified risks in 2014 and through a more in depth look at items that were included in the Parking Lot in 2014.

The secondary objective of the workshop was to, for any unacceptable residual risks identified in the ranking exercise, identify potential mitigation measures that would further reduce the overall risk ranking and bring the residual risks into a more acceptable location in the risk matrix. These potential mitigations could then be incorporated into the planning work being undertaken currently for the development of the RCP v2018-01.

2.2 SCOPE AND APPROACH

The FMEA covered Minto's RCP v2017-01 and recent mine plan changes (e.g. inclusion of Ridgetop Pits and Dump) and related design elements for the proposed closure measures. This included RCP elements that were not fully developed in 2014 (e.g. passive water treatment plan, waste dump re-grading plans.) The FMEA was an 'internal' exercise (attended only by Minto management and engineering staff and by relevant engineering and planning consultants) and was focused only on specific consequence types related to traditional engineering and design related-risks and consequences. Further details on the risk rating tools are provided in Section 2.4 and Appendix A.

The FMEA was carried out by revisiting the relevant risk register components and entries from 2014, and modifying the language of the risk/consequence statements and the risk rankings if and where necessary. Risk/consequence statements were added/modified/removed as required.

The FMEA was conducted both on a "Facility" basis (separate risk registers for each type of facility) and on topics that are appropriately addressed on a "Site-Wide" basis. These are generally reflective of the reclamation and closure measures presented in RCP Section 7, and included:

Facilities:

- Open Pits
- Waste Rock Dumps
- Dry Stack Tailings Storage Facility

Site-Wide Topics:

- Water Quality/Water Treatment

- Water Conveyance
- Administrative

2.3 WORKSHOP STRUCTURE

2.3.1 OCTOBER WORKSHOP

The first FMEA workshop was held at SRK Consulting's Vancouver office on October 18th and 19th, 2017. The workshop was facilitated by S. Keesey of AEG. The two-day workshop included participation by representatives of Minto, Alexco Environmental Group, and SRK Consulting. The topics rated in the October workshop included Pits, Waste Rock Dumps, Water Conveyance and Administrative.

The participants are listed in Table 2-1.

Table 2-1: October FMEA Workshop Participants

Name	Company
Ryan Herbert	Minto Explorations Ltd.
Kevin Cymbalisty	Minto Explorations Ltd.
Scott Keesey	Alexco Environmental Group
Dylan MacGregor	SRK Consulting
Erik Ketilson	SRK Consulting
Sarah Portelance	SRK Consulting
Peter Mikes	SRK Consulting

2.3.2 DECEMBER WORKSHOP

The second FMEA workshop was held at SRK Consulting's Vancouver office on December 1st, 2017 and was again facilitated by S. Keesey of AEG. The one-day workshop included participation by representatives of Minto, Contango Strategies, Alexco Environmental Group, and SRK Consulting. The December workshop was focused on Water Treatment and Water Quality aspects of Minto's RCP.

The participants are listed in Table 2-2.

Table 2-2: December FMEA Workshop Participants

Name	Company
Ryan Herbert	Minto Explorations Ltd.
Monique Simair	Contango Strategies Ltd.
Scott Keesey	Alexco Environmental Group
Soren Jensen	SRK Consulting
Dylan MacGregor	SRK Consulting

2.4 FMEA PROCESS

The FMEA was carried out using a consequence-likelihood method, utilizing three risk rating tools (the consequence-severity matrix, the likelihood chart, and the risk matrix) which are included in Appendix A. These tools were identical to those utilized by the multi-stakeholder group in 2014 with one exception. As this was in internal risk ranking exercise and was intended to focus primarily on engineering and design aspects of the project, the risk evaluation was limited to the three most relevant consequence types: Environmental Impact, Cost Consequence, and Human Health and Safety. Minto expects to re-visit the FMEA process as a multi-stakeholder process (as a condition of an amended Water Use Licence) and will integrate the remaining consequence types into that exercise with the appropriate stakeholders in attendance. Results of this additional exercise will be included in future versions of the Minto RCP.

The 2014 risk register was simplified in advance by removing the risk 'category' indicator (physical stability, chemical stability, etc.) from the tables and the coding. These generic closure objective considerations were still utilized to organize thoughts and risk types. The revised risk register was circulated in advance to participants of the sessions for familiarity to make the workshop more efficient.

2.4.1 REMOVAL OF SELECT 2014 FMEA RISK STATEMENTS

A select number of risk/consequence statements from the 2014 FMEA were removed from the 2017 register and were not reviewed during the FMEA process. The majority of the entries removed from the register were entries in the consequence categories that were not evaluated (Traditional Land Use, Community/Media/Reputation, and Regulatory/Legal). The remaining removed entries were removed only if the underlying aspects of the entry, such as the mine plan, were no longer relevant.

2.4.2 RISK RATING PROCESS

The remaining risk/consequence statements from the 2014 FMEA were reviewed and the likelihood and consequence for each were determined using the consequence-severity and likelihood tools to reach a consensus risk rating. Additionally, new risk/consequence statements were developed by the group, with their risk rating determined in the same manner. The risk ratings were recorded in a risk register spreadsheet that was projected on a screen for participants to refer to and provide feedback on during the meeting. The resultant risk IDs were compiled into risk matrices within the Excel workbook with each risk ID placed in the appropriate cell for its risk rating. An individual risk matrix was developed for each facility and/or topic, as well as a summary risk matrix containing all the risk IDs.

2.4.3 RE-RANKING PROCESS

For a few select risk/consequence statements, the likelihood-consequence ranking was revisited after the inclusion of mitigative measures that are in addition to those included in RCP v2017-01. In some instances, these are measures which were already planned for inclusion in RCP v2018-01. In some other instances, the initial ranking warranted inclusion of different or additional measures (i.e. the risk ranking was unacceptably high.) This 're-ranking' effort was captured in a separate set of matrices (see Section 3) for comparison to the initial risk matrices.

3. FMEA RESULTS

The results of the 2017 FMEA update are described and referenced in the following sections, including the Risk Register, summary and detailed Risk Matrices, risks which were not re-evaluated in 2017 (and rationale) and how 2014 'Parking Lot' items were addressed.

3.1 RISK REGISTERS (APPENDIX B)

These register tables B-1 through B-6 in Appendix B capture following for each topic evaluated:

- the risk and consequence statements with an ID number and consequence category;
- the likelihood and consequence ranking assigned by the team;
- the resultant risk classification for the likelihood and consequence ranking;
- an indication if the risk/consequence statement was changed from the 2014 entry, or if it is a new entry;
- comments that identify assumptions or clarifications regarding the ranking selected; and
- if required – new potential mitigations and 're-ranking' of the mitigated risk.

The ID number of each risk/consequence statement is located on the overall risk matrix, as outlined below in section 3.2. The registers are the most detailed product of the FMEA exercise.

3.2 RISK MATRICES (APPENDIX C)

These 'heat maps' illustrate the overall outcomes of the FMEA ranking exercise. Each risk/consequence statement is placed by its ID number on the matrix in the appropriate risk classification location based on the likelihood and consequence rating. The key to the matrix (Figure 1) and the final 'heat maps' of the exercise (Figures 2 and 3) are presented below, and the full matrices – including those for the individual Topics for both the current (v2017-01) and expected (v2018-01) suites of mitigation measures associated with those RCPs – are provided in Appendix C.

Likelihood	Consequence Severity				
	Very Low	Minor	Moderate	Major	Critical
Almost Certain	Moderate	Moderately High	High	Very High	Very High
Likely	Moderate	Moderate	Moderately High	High	Very High
Possible	Low	Moderate	Moderately High	High	High
Unlikely	Low	Low	Moderate	Moderately High	Moderately High
Very Unlikely	Low	Low	Low	Moderate	Moderately High

Figure 1: Risk Matrix 'Key' with identification of the risk classifications that result from likelihood and consequence ratings.

Likelihood							
	Almost Certain	5		D4		D27	
	Likely	4		WT18, WT32, W10	WT38, WT42, W5		
	Possible	3	P16, P17, P27, WT23, W11	P14, T2, T3, D5, D12, D14, D18, D19, D29, D30, D30, WT11, WT29, WT43, W22, W23, W24, W32	P18, WT21, A4	A5, A6	
	Unlikely	2	P1, P19, P21, P22, D10, D24, W29, W30	P8, P9, P11, P12, T1, T7, T11, T12, T13, D6, D11, D13, D20, D31, D31, WT2, WT5, WT8, WT14, WT28, WT30, WT31, WT35, WT36, WT40, W6, W14, W27, W33, W36, W37, A3	WT1, WT4, WT6, WT7, WT9, WT10, WT13, WT15, WT19, WT22, WT34, W28	A1	WT3, WT12
	Very Unlikely	1	W3, W4, W15, W17	P23, T4, T8, T9, D23, D26, WT17, WT25, WT26, WT27, WT37, WT41, W1, W2, W8, W16, W18, W19, W20	P5, P6, P10, P15, P20, P24, P25, P26, T10, D1, D2, D3, D7, D8, D9, D15, D16, D17, D21, D22, D25, WT16, WT20, WT24, WT33, WT39, WT44, W7, W9, W13, W21, W31	T5, T6, D28, W12, A8	P7, P13, A7
			1	2	3	4	5
			Very Low	Minor	Moderate	Major	Critical
			Consequence				

Figure 2: Overall summary risk matrix from 2017 FMEA for RCP v2017-01.

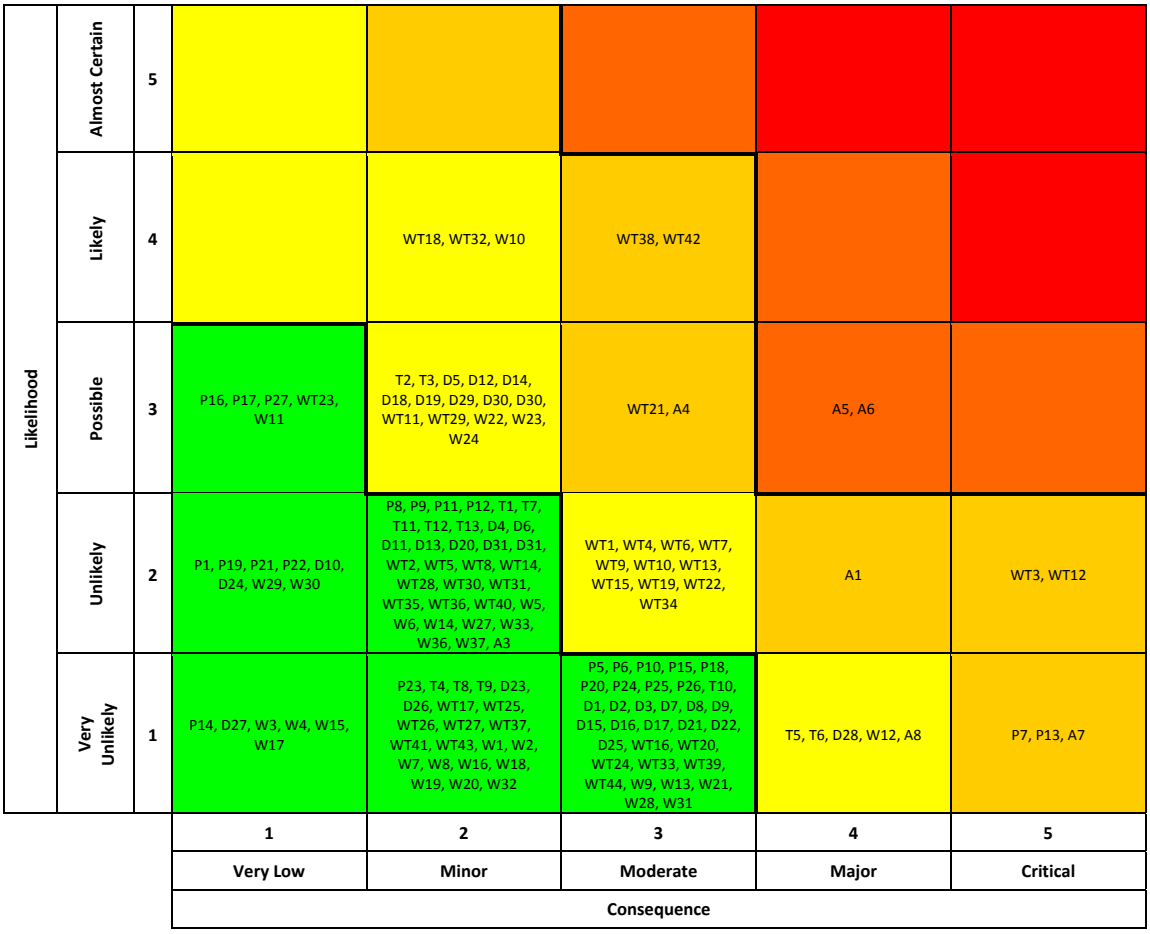


Figure 3: Overall summary risk matrix from 2017 FMEA for RCP v2017-01 with inclusion of additional proposed mitigation measures being planned for inclusion in RCP v2018-01.

Overall, the exercise returned mostly 'low' and 'moderate' risks associated with the v2017-01 plan. The unacceptable risk areas (high and very-high) contained only three entries. Two of the three were Administrative risk/consequence statements that were not changed from 2014 (although the 2017 workshop team strongly disagrees with the previous risk rating). Further, the 2017 team suggests that the risk has either been mitigated since 2014 (with additional financial security posted with YG) or there is no conceivable way to mitigate the risk, as it assumes violation of future licence conditions. Both entries are of the Cost Consequence category.

The third unacceptable risk statement regarding a failure to relocate SAT material prior to completion of the Main Pit Dump (as with some others) was subject to a 're-ranking' after the identification of additional mitigation measures, which reduced the risk classification from 'very high' to 'low'.

3.3 RISKS NOT RE-EVALUATED (APPENDIX D)

Table D-1 (in Appendix D) includes 2014 register entries that were not re-evaluated. 2014 comments regarding the risk statements were retained, and rationale for the entry not being re-evaluated in 2017 is included. The most common reasons for not re-evaluating entries were:

- The consequence category was not utilized in the 2017 workshop;
- The entry was expanded into a number of more detailed risk statements;
- The entry was a carry over from 2013 and/or is addressed elsewhere and is redundant; and
- The entry pertained to an element of the mine plan or RCP that is no longer relevant (e.g. Main Dam).

The 2017 FMEA team acknowledges that some of these risks are still worthy of re-evaluation, yet is confident that the 2017 RCP addresses most of the remaining higher classified risk statements, particularly from categories that were not evaluated in the 2017 workshop. For example, the highest classified risks in this table relate to consequences from a general failure to conduct preventative maintenance and corrective actions leading to system failures. The likelihood of these failures was rated as 'LIKELY' in 2014 on the basis of the duration of post-closure monitoring and maintenance in RCP 2014-01. RCP v2017-01 includes a plan and a closure (security) cost allowance for a much longer PCMM duration.

3.4 PARKING LOT ITEMS

The 2014 issues raised that were recorded in the Parking Lot were itemized previously in Section 1.2. They are reproduced again below with an indication of how each item was addressed in RCP v2017-01 and/or the 2017 FMEA.

- **Current closure plan is deficient with respect to showing final reclaimed facilities, toes of re-graded slopes and location of secondary and tertiary water conveyance;**

  The designs for final reclaimed facilities have been advanced substantially in RCP v2017-01, and the final re-grading plans for the waste facilities were also available for the 2017 FMEA and will be included in RCP v2018-01.

- **More information requested regarding the status of the Reclamation Research Plan and the Main Waste Dump revegetation trials;**

  RCP v2017-01 included additional information regarding the RRP, particularly related to the Constructed Wetland Treatment System research. RCP v2018-01 further updates work on this research component (Section 2.2) and the accompanying cover design report (Updated Closure Cover Design for the Minto Mine 2018 Reclamation and Closure Plan (SRK Consulting, January 2018)) incorporates recommendations based on observations of the Main Waste Dump revegetation trials during inspections carried out during July 2017.

- **Trafficability layer is required over the Ridgetop North Pit tailings backfill whereas costing only allows for 0.5m of overburden;**

  The Ridgetop Pits were not a part of the mine plan that underpinned RCP v2017-01. Now that the Ridgetop pits have been re-integrated into the mine plan, a trafficability layer over the Ridgetop North Pit tailings backfill has been added to the mitigative measures for that facility in RCP v2018-01.

- **Need to advance the discussion/determination of what constitutes "reasonable and practicable" passive treatment, establish protocols and revisit the options evaluation;**

  RCP v2017-01 presented the documentation of the systematic process that was undertaken to identify and select appropriate passive treatment technologies for the Minto Site. The RCP also presented a detailed design for the installation of a constructed wetland treatment system as part of the closure configuration, with associated

monitoring, maintenance and adaptive management plans to ensure its efficacy in the post-closure period. The system was evaluated in depth in the Water Treatment topic in the 2017 FMEA workshop, which included a select group of technical experts who were responsible for the treatment system design.

- **SFN reiterated their concern that the consequence category of "Community/Media/Reputation" is biased because SFN are lumped together with groups having other interests and perspective; and**

  This consequence category was not included in the 2017 workshop exercise.

- **The current closure plan does not sufficiently address signage and access control. There is a need to retain institutional controls and maintain signage in perpetuity.**

  This issue has not been addressed in any RCP updates. It continues to be a future land use management decision requiring resolution with SFN.

4. SUMMARY

The risk registers updated during the 2017 FMEA workshops are provided in Appendix B and the risk matrices, both the initial (RCP v2017-01) and re-ranked (including proposed mitigations in RCP v-2018-01), are provided in Appendix C.

Although some risk consequence categories utilized in previous FMEA sessions were not included in this exercise, previous risk entries were evaluated in a more detailed fashion, leading to more registry entries than in the 2014 exercise. A more detailed and developed RCP, including a detailed plan for construction and operation of a passive treatment installation, allowed a fulsome evaluation of detailed risk statements, including risks and consequences of failure of the passive water treatment system. The 2017 FMEA addressed items from the 2014 FMEA 'Parking Lot'.

Overall, the exercise returned mostly 'low' and 'moderate' risks associated with the v2017-01 plan. The unacceptable risk areas (high and very-high) contained only three entries. Two of the three were Administrative risk/consequence statements that were not changed from 2014 (although the 2017 workshop team strongly disagrees with the previous risk rating). Further, the 2017 team suggests that the risk has either been mitigated since 2014 (with additional financial security posted with YG) or there is no conceivable way to mitigate the risk, as it assumes violation of future licence conditions. Both entries are of the Cost Consequence category.

The third unacceptable risk statement regarding a failure to relocate SAT material prior to completion of the Main Pit Dump (as with some others) was subject to a 're-ranking' after the identification of additional mitigation measures, which reduced the risk classification from 'very high' to 'low'.

APPENDIX A.
Risk Rating Tools

Consequence-Severity Matrix

	Severity Descriptors				
Consequence Categories	**Very Low**	**Minor**	**Moderate**	**Major**	**Critical**
1. Environmental Impact (EI)	No impact.	Minor localized or short-term impacts.	Significant impact on valued ecosystem component.	Significant impact on valued ecosystem component and medium-term impairment of ecosystem function.	Serious long-term impairment of ecosystem function.
2. Consequence Costs (CC)	< $100,000	$100,000 - $500,000	$ 500,000 - $2.5 Million	$2.5-$10 Million	>$10 Million
3. Human Health and Safety (HHS)	Low-level short-term subjective symptoms. No measurable physical effect. No medical treatment.	Objective but reversible disability/impairment and /or medical treatment injuries requiring hospitalization.	Moderate irreversible disability or impairment to one or more people.	Severe irreversible disability or impairment to one or more people.	Single fatality or multiple fatalities.

Likelihood Terminology

Likelihood	Frequency Descriptor 1	Frequency Descriptor 2
Almost Certain	Happens often	High frequency (more than once every 5 years)
Likely	Could easily happen	Event does occur, has a history, once every 15 years
Possible	Could happen and has happened elsewhere	Occurs once every 40 years
Unlikely	Hasn't happened yet but could	Occurs once every 200 years
Very Unlikely	Conceivable, but only in extreme circumstances	Occurs once every 1000 years

APPENDIX B.

2017 FMEA Risk Registers

TABLE B-1: Open Pits Risk Register

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Risk
Low
Moderate
Moderately High
High
Very High

Open Pits (P)										Mitigated		
RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING L	RISK RATING C	CLASSIFICATION	changed?	COMMENTS	Potential Mitigations	RISK RATING L	RISK RATING C	CLASSIFICATION
P1	Area 2 Pit	Precipitation higher than design asssumption (1:200 yr 24 hr)	Damage to outlet structure	CC	Unlikely	Very Low	Low	Y	Assumes earthwork repairs to spillway Plan includes cost allocation for regular maintenance				
P2	Area 2 Pit	Precipitation higher than design asssumption (1:200 yr 24 hr)	Erosion of Ditch 400 channel and damage to toe of Main Dam, leading to breach of dam and release of tailings to lower Minto Creek	EI				Y	no Main Dam, not relevant				
P3	Area 2 Pit	Precipitation higher than design asssumption (1:200 yr 24 hr)	Erosion of Ditch 400 channel and damage to toe of Main Dam, leading to breach of dam and release of tailings to lower Minto Creek	CC				Y	no Main Dam, not relevant				
P4	Area 2 Pit	Precipitation higher than design asssumption (1:200 yr 24 hr)	Erosion of Ditch 400 channel and damage to toe of Main Dam, leading to breach of dam and release of tailings to lower Minto Creek	HHS				Y	no Main Dam, not relevant				
P5	Area 2 Pit	Pit wall failure in Area 2	Wave of water released from pit causing damage to downstream facilities (ditches, passive treatment system, covers)	EI	Very Unlikely	Moderate	Low	Y	Large degree of uncertainty regarding the likelihood of failure duration limited to 'seconds' A2 Pit outflow would not pass all of wave, wave would disperse down conveyance, high flow bypass would protect wetlands				
P6	Area 2 Pit	Pit wall failure in Area 2	Wave of water released from pit causing damage to downstream facilities (ditches, passive treatment system, covers)	CC	Very Unlikely	Moderate	Low	Y	includes significant repairs to all downstream structures, including CWTS				
P7	Area 2 Pit	Pit wall failure in Area 2	Wave of water released from pit and causes a fatality.	HHS	Very Unlikely	Critical	Moderately High	N	Same as above; -Ranking preliminary; Failure has not been evaluated. -If a lower likelyhood option was available, it would have been selected.				
P8	Area 2 Pit	Ongoing overburden south wall erosion (A2S3) from either wave action or runoff	sediment deposition in downstream conveyance structures leading to increased maintenance requirements beyond those planned	CC	Unlikely	Minor	Low	NEW	assumes mitigation by armouring would be implemented				
P9	Area 2 Pit	Ongoing overburden south wall erosion (A2S3) from either wave action or runoff	sediment deposition in downstream conveyance structures leading to unacceptable TSS concentrations in site discharge	EI	Unlikely	Minor	Low	NEW	TSS settlment in pit would mitigate most of this potential				
P10	Area 2 Pit	Pit water quality degraded unacceptably (dissolved constituents)	Unacceptable water quality conditions for water fowl / wildlife	EI	Very Unlikely	Moderate	Low	Y	Scenario was evaluated (Minnow)				
P11	Minto North Pit	Positive water balance for pit	Overtopping or piping and erosion at spill point and sediment release downstream, causing unacceptable sedimentation in Mcginty Creek	EI	Unlikely	Minor	Low	N	This is included in Closure AMP. Observations during operations and Post-closure 1 period will be useful in better understanding future fate of MN Pit water balance				
P12	Minto North Pit	Massive rapid pit wall failure into pit full of water	Wave of water spilling over rim of pit causing erosion and downstream sedimentation and riparian damage	EI	Unlikely	Minor	Low	Y	consideration was given to the fact that there has already been a pit wall failure in Minto North pit, so likelihood of combined failure mode and consequence was changed from very unlikely to unlikely				
P13	Minto North Pit	Pit wall failure	Causes a fatality.	HHS	Very Unlikely	Critical	Moderately High	NEW	mitigated by SWP for monitoring in pit				
P14	Ridgetop North Pit	Erosion of downstream slope due to runoff from covered tailings	Need for repairs	CC	Possible	Minor	Moderate	N	Could be mitigated through AMP or addressed in closure plan	add route for surface water drainage	Very Unlikely	Very Low	Low
P15	Ridgetop North Pit	Settlement of tailings	Ponding on surface of covered tailings resulting in increased infiltration and leading unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	N					
P16	Ridgetop North Pit	High wall slope failure	Cover damage, exposure of tailings and need for cover repairs	CC	Possible	Very Low	Low	Y	pit wall failures have been observed on site, but cost of repairs are modest				
P17	Ridgetop North Pit	Seepage of water within RNPTMF against E/NE overburden wall	Slope instability in overburden in A2S3 and need for repairs	CC	Possible	Very Low	Low	Y	conservative slope stability analysis to evaluate risk of this failure mode was undertaken, possible sloughing in short term, repairs are modest				
P18	Ridgetop North Pit	Thawing of entrained ice in tailings	Cover damage, exposure of tailings and need for cover repairs	CC	Possible	Moderate	Moderately High	NEW		will be mitigated through appropriate tailings deposition plan	Very Unlikely	Moderate	Low

TABLE B-1: Open Pits Risk Register

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Risk
Low
Moderate
Moderately High
High
Very High

Open Pits (P)										Mitigated		
RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING L	RISK RATING C	CLASSIFICATION	changed?	COMMENTS	Potential Mitigations	RISK RATING L	RISK RATING C	CLASSIFICATION
P19	Ridgetop North Pit	Thawing of entrained ice in tailings leads to ponding and saturation	Slope instability in overburden in A2S3 and need for repairs	CC	Unlikely	Very Low	Low	Y	conservative slope stability analysis to evaluate risk of this failure mode was undertaken, possible sloughing in short term, repairs are modest				
P20	Ridgetop North Pit	piping of tailings material or internal erosion of OVB below RTNTF	instability and uncontrolled release of tailings and unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	NEW	scenario was assessed in response to information request during YWB review of Phase V/VI Expansion Project				
P21	Ridgetop North Pit	piping of tailings material through cover results in 'boils' of exposed tailings	need for repairs/clean up of exposed tailings	CC	Unlikely	Very Low	Low	NEW					
P22	Main Pit	Precipitation higher than design asssumption (1:200 yr 24 hr)	Damage to outlet structure	CC	Unlikely	Very Low	Low	NEW	Assumes earthwork repairs to spillway Plan includes cost allocation for regular maintenance				
P23	Main Pit	Pit wall failure	Wave of water released from pit causing damage to downstream facilities (ditches, passive treatment system, covers)	CC	Very Unlikely	Minor	Low	NEW	assuming less than 2m of water cover, so 'wave potential' not the same as A2 pit, lower volume of water, less damage downstream				
P24	Main Pit	Pit wall failure	Wave of water released from pit causing damage to downstream facilities (ditches, passive treatment system, covers)	EI	Very Unlikely	Moderate	Low	NEW					
P25	Main Pit	Pit water quality degraded unacceptably (dissolved constituents)	Unacceptable water quality conditions for water fowl / wildlife	EI	Very Unlikely	Moderate	Low	NEW	Scenario was evaluated (Minnow)				
P26	Main Pit	Wind suspension of tailings mobilizes tailings material out of MPTMF	sediment deposition in downstream conveyance structures leading to increased maintenance requirements beyond those planned	CC	Very Unlikely	Moderate	Low	NEW	Traffic surface element of tailings suface cover will extend below final water elevation to mitigate wind wave turbulence circle suspending tailings				
P27	Main Pit	piping of tailings material through cover results in 'boils' of exposed tailings	need for repairs/clean up of exposed tailings	CC	Possible	Very Low	Low	NEW					

TABLE B-2: DSTSF Risk Register

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Dry Stack Tailings Storage Facility (T)											Mitigated		
RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING		CLASSIFICATION	changed?	COMMENTS	Potential Mitigations	RISK RATING		CLASSIFICATION
					L	C					L	C	
T1	DSTSF/MVFE	Long-term permafrost degradation underlying the DSTSF leads differential settlement and ponding of water resulting in increased infiltration leading to	Unacceptable downstream water quality	EI	Unlikely	Minor	Low	N					
T2	DSTSF/MVFE	Long-term permafrost degradation underlying the DSTSF leads to differential settlement and ponding of water	Need for repair	CC	Possible	Minor	Moderate	N					
T3	DSTSF/MVFE	Long-term permafrost degradation underlying the DSTSF leads differential settlement and ponding of water	Erosion of cover materials and sediment loading to passive treatment system	EI	Possible	Minor	Moderate	N	Maintenance issue				
T4	DSTSF/MVFE	Re-initiation of shear zone due to thawing of permafrost from the bottom up due to high excess pore pressure and movement in the cross- valley direction, leading to cracking of the cover and increased infiltration	Impacts to downstream water quality.	EI	Very Unlikely	Minor	Low	N	Mitigated by construction of MVFE S2				
T5	DSTSF/MVFE	Deep-seated downvalley slope failure of the MVFE (Section F)	Instability of the DSTSF and cracking of the cover, failure of conveyance ditches and need for additional stabilization measures.	CC	Very Unlikely	Major	Moderate	N	MVFE S2 design report calculates a minimum FOS of 2.3. MVFE S2 has been constructed years before closure, and will continue to produce years of performance monitoring data. No significant concerns with down valley movement with one year of data.				
T6	DSTSF/MVFE	MVFE S2 does not extend far enough downvalley	Cross-valley movement at the eastern extent of the current movement affecting the DSTSF and need for additional stabilization measures.	CC	Very Unlikely	Major	Moderate	NEW	MVFE S2 has been constructed years before closure, and will continue to produce years of performance monitoring data. No significant concerns with down valley movement with one year of data.				
T7	DSTSF/MVFE	Toe failure of the MVFE	Instability and cracking of the cover, failure of conveyance ditches and need for additional stabilization measures.	CC	Unlikely	Minor	Low	Y	Facility is constructed, and no concerns based on performance data.				
T8	DSTSF/MVFE	Toe failure of the MVFE	failure of conveyance ditch and mobilization of sediment, leading to unacceptable TSS concentrations in site discharge	EI	Very Unlikely	Minor	Low	NEW	Facility is constructed, and no concerns based on performance data.				
T9	DSTSF/MVFE	Movement (lateral movement or differential settlement) reduces or blocks flow from the finger drains underlying the DSTSF	Increased pore pressure within tailings mass, raising of water table within tailings mass and ultimately increased daylighting of groundwater upgradient of DSTSF	EI	Very Unlikely	Minor	Low	N					
T10	DSTSF/MVFE	Facility instability results in waste material exposure to water and reduced cover performance	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	NEW					
T11	DSTSF/MVFE	Erosion on slopes leads to loss of cover	Need for repairs beyond regular maintence planned	CC	Unlikely	Minor	Low	NEW					
T12	DSTSF/MVFE	Die-back of cover vegetation after successful establishment and acceptance leads to erosion	Need for repairs beyond regular maintence planned	CC	Unlikely	Minor	Low	NEW					
T13	DSTSF/MVFE	Root throw results in increased infiltration over the long term	Need for repairs beyond regular maintence planned	CC	Unlikely	Minor	Low	NEW					

TABLE B-3: Waste Rock Dump Risk Register

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Risk
Low
Moderate
Moderately High
High
Very High

Waste Rock Dumps (D)										Mitigated		
RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING L	RISK RATING C	CLASSIFICATION	changed?	COMMENTS	Potential Mitigations	RISK RATING L	RISK RATING C	CLASSIFICATION
D1	Main Waste Dump	Run-on water from upgradient catchment of MWD increases flow subsurface and contaminant loading from waste rock	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	N	MWD has limited upgradient catchment area and limited resulting run-on				
D2	Main Waste Dump	Ponding of water on surface of MWD leads to excessive infiltration, increases flow subsurface and contaminant loading from waste rock	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	Y	has been mitigated by shaping surface to limit ponding				
D3	Main Waste Dump	Instability results in waste material exposure to water	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	N	Likelihood supported by stability evaluations in MWDE design report				
D4	Main Waste Dump	Erosion on steeper portion of MWD leads to loss of cover	Need for repairs beyond regular maintence planned	CC	Almost Certain	Minor	Moderately High	Y	has been observed at site on this facility	Waste Rock wrap at toe to reduce slope of bottom of slopes	Unlikely	Minor	Low
D5	Main Waste Dump	Die-back of cover vegetation after successful establishment and acceptance leads to erosion	Need for repairs beyond regular maintence planned	CC	Possible	Minor	Moderate	N	MWD has been covered/ planted for several years. Appropriate selection of veg species would reduce chance of wholesale die-back. Can be mitigated through design and/or O&M plan. Three additional years of monitoring observations of MWD revegetation trials have not shown any signs of die-back, however likelihood was kept the same.				
D6	Main Waste Dump	Root throw results in increased infiltration over the long term	Need for repairs beyond those planned	CC	Unlikely	Minor	Low	N					
D7	Southwest Waste Dump	Run-on water from upgradient catchment of SWD increases flow subsurface and contaminant loading from waste rock	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	N	SWD has limited upgradient catchment area and limited resulting run-on				
D8	Southwest Waste Dump	Ponding of water on surface of SWD leads to excessive infiltration, increases flow subsurface and contaminant loading from waste rock	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	Y	Requires failure of AMP to realize the scenario. Reclamation plan now includes shaping surface to limit ponding - likelihood reduced; maintenance may be required to restore contouring (if differential settlement occurs) to prevent ponding. There is a BGM cover on HGW, so ponding+increased infiltration is less likely for HGW				
D9	Southwest Waste Dump	Instability results in cover compromised and waste material exposure to water	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	N	Likelihood supported by stability evaluations in SWD design report. Foundation includes permafrost overburden, but the design considered the existing foundation conditions.				
D10	Southwest Waste Dump	Instability results in cover compromised	Need for repairs beyond regular maintence planned	CC	Unlikely	Very Low	Low	NEW	Planned maintenance most likely to cover this type of repair work.				
D11	Southwest Waste Dump	Erosion on steeper portions of SWD leads to loss of cover	Need for repairs beyond regular maintence planned	CC	Unlikely	Minor	Low	Y	Likelihood can be reduced through implementation of design, including planned vegetation, and planned maintenance is most likely to cover this type of repair work.				
D12	Southwest Waste Dump	Die-back of cover vegetation after successful establishment and acceptance leads to erosion	Need for repairs beyond regular maintence planned	CC	Possible	Minor	Moderate	N	SWD will have been covered/ planted for several years; appropriate selection of veg species would reduce chance of wholesale die-back. Can be mitigated through design and/or O&M plan. Expect experience from MWD to be replicated.				
D13	Southwest Waste Dump	Root throw results in increased infiltration over the long term	Need for repairs beyond regular maintence planned	CC	Unlikely	Minor	Low	N					
D14	Southwest Waste Dump	Root throw results in damage to engineered cover (BGM) over HGW leads to increased infiltration over the long term	Need for repairs beyond regular maintence planned	CC	Possible	Minor	Moderate	N		HGW will be removed prior to closure, area treated similar to rest of SWD facility, therefore this risk is eliminated (not ranked.)			
D15	Ridgetop Waste Dump	Run-on water from upgradient catchment of RWD increases flow subsurface and contaminant loading from waste rock	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	N	MWD has limited upgradient catchment area and limited resulting run-on				
D16	Ridgetop Waste Dump	Ponding of water on surface of MWD leads to excessive infiltration, increases flow subsurface and contaminant loading from waste rock	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	Y	has been mitigated by shaping surface to limit ponding				
D17	Ridgetop Waste Dump	Instability results in waste material exposure to water	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	N	Likelihood supported by stability evaluations in MWDE design report				
D18	Ridgetop Waste Dump	Erosion on steeper portion of MWD leads to loss of cover	Need for repairs beyond regular maintence planned	CC	Possible	Minor	Moderate	N	Can be mitigated through design and/or O&M plan				

TABLE B-3: Waste Rock Dump Risk Register

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Risk
Low
Moderate
Moderately High
High
Very High

Waste Rock Dumps (D)										Mitigated		
RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING L	RISK RATING C	CLASSIFICATION	changed?	COMMENTS	Potential Mitigations	RISK RATING L	RISK RATING C	CLASSIFICATION
D19	Ridgetop Waste Dump	Die-back of cover vegetation after successful establishment and acceptance leads to erosion	Need for repairs beyond regular maintence planned	CC	Possible	Minor	Moderate	N	RWD will have been covered/ planted for several years; appropriate selection of veg species would reduce chance of wholesale die-back. Can be mitigated through design and/or O&M plan				
D20	Ridgetop Waste Dump	Root throw results in increased infiltration over the long term	Need for repairs beyond those planned	CC	Unlikely	Minor	Low	N					
D21	Reclamation OVB, Ridgetop South & Area 118 Backfill Dumps	Run-on water from upgradient catchments increases flow subsurface and contaminant loading from overburden	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	N	Dump contents are overburden				
D22	Reclamation OVB, Ridgetop South & Area 118 Backfill Dumps	Thawing of ice-rich OVB leads to settlement and ponding of water on surface of overburden dumps leads to excessive infiltration, increases flow subsurface and contaminant loading from overburden	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	N	Designs of A118 and RS BD are mounded to shed water.				
D23	Reclamation OVB, Ridgetop South & Area 118 Backfill Dumps	RS BD is not constructed to design limits, resulting in development of a pit lake within RS BD pit leading to spill of pit water and erosion of downgradient slope	Need for development of channel	CC	Very Unlikely	Minor	Low	N	Likely will not form lake based on lack of water encountered in A118 pit. Could be mitigated by filling with waste rock during mining or by filling later with ob or waste rock				
D24	Reclamation OVB, Ridgetop South & Area 118 Backfill Dumps	Die-back of re-vegetation after successful establishment and acceptance leads to erosion	Need for repairs beyond regular maintence planned	CC	Unlikely	Very Low	Low	Y	No requirement to maintain cover integrity due to dump material (OVB); have observed natural revegetation on portions of ROD				
D25	Site Wide - Covers	Burrowing animals burrow in covers, increasing infiltration	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	NEW					
D26	Site Wide - Covers	Burrowing animals burrow in covers and/or trail footprints leading to erosion	Need for repairs beyond regular maintence planned	CC	Very Unlikely	Minor	Low	NEW					
D27	Main Pit Dump/SAT Dump	SAT material is not relocated prior to completion of Main Pit Dump,	portion of MPD needs to be excavated for SAT material relocation	CC	Almost Certain	Major	Very High	NEW		Main Pit Dump **has been revised** to not cover SAT material, and construction is underway	Very Unlikely	Very Low	Low
D28	Main Pit Dump/SAT Dump	Construction of MPD leads to re-activation of South Wall shear zone and failure of SouthWall/MPD into Main Pit	need for re-stabilization and repairs to cover and conveyance channels	CC	Very Unlikely	Major	Moderate	NEW					
D29	Main Pit Dump/SAT Dump	Erosion on steeper portion of MWD leads to loss of cover	Need for repairs beyond regular maintence planned	CC	Possible	Minor	Moderate	NEW	Can be mitigated through design and/or O&M plan				
D30	Main Pit Dump/SAT Dump	Die-back of cover vegetation after successful establishment and acceptance leads to erosion	Need for repairs beyond regular maintence planned	CC	Possible	Minor	Moderate	NEW	MPD will have been covered/ planted for several years; appropriate selection of veg species would reduce chance of wholesale die-back. Can be mitigated through design and/or O&M plan				
D31	Main Pit Dump/SAT Dump	Root throw results in increased infiltration over the long term	Need for repairs beyond those planned	CC	Unlikely	Minor	Low	NEW					

TABLE B-4: Water Conveyance Structure Risk Register

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Risk
Low
Moderate
Moderately High
High
Very High

Water Conveyance (W)											Mitigated		
RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING L	RISK RATING C	CLASSIFICATION	changed?	COMMENTS	Potential Mitigations	RISK RATING L	RISK RATING C	CLASSIFICATION
W1	Ditch A	Instability of the MPD or South Wall compromises Ditch A leads to exposure of waste rock and infiltration	Unacceptable water quality downstream	EI	Very Unlikely	Minor	Low	Y	Channel no longer runs on top of MPD				
W2	Ditch A	Instability of the MPD or South Wall compromises Ditch A	Repair requirements	CC	Very Unlikely	Minor	Low	Y					
W3	Ditch A	thaw consolidation of permafrost overburden in South Wall leads to deformation of Ditch A	Repair requirements	CC	Very Unlikely	Very Low	Low	NEW	Design tolerates thaw consolidation				
W4	Ditch A	Flows exceed Ditch A capacity resulting in failure of water conveyance structures	Repair requirements	CC	Very Unlikely	Very Low	Low	NEW	Alignment follows new valley bottom landform				
W5	Ditch B	Thaw degradation leads to retrogressive failure at inlet of A2S3 Pit	Need for repairs beyond regular maintenance already planned	CC	Likely	Moderate	Moderately High	NEW	2016 design does not account for consequences of long term thaw	cost allowance for design of improved inlet structure	Unlikely	Minor	Low
W6	Ditch B	Thaw degradation leads to retrogressive failure at inlet of A2S3 Pit	sediment deposition in downstream conveyance structures leading to unacceptable TSS concentrations in site discharge	EI	Unlikely	Minor	Low	NEW	TSS settlment in pit would mitigate most of this potential				
W7	Ditch B	Flows exceed Ditch B capacity resulting in failure of water conveyance structure and flows over DSTSF leading to cover damage	Repair requirements	CC	Very Unlikely	Moderate	Low	NEW		DSTSF Landform design includes regrading to direct flows around DSTSF rather than over	Very Unlikely	Minor	Low
W8	TDD	Extreme event leads to failure of TDD conveyance structure upgradient of DSTSF, flow onto DSTSF leading to cover damage	Repair requirements	CC	Very Unlikely	Minor	Low	NEW	discussion considered whether tailings could be mobilized, was decided not likely enough to consider				
W9	TDD	Leakage from TDD conveyance structure upgradient of DSTSF increases flow subsurface and contaminant loading from tailings	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	Y	updgraded TDD constructed				
W10	Primary Channels	Winter ice development in channels causes freshet flows to be above protected elevations, leading to erosion	Need for repairs beyond regular maintenance already planned	CC	Likely	Minor	Moderate	Y	May be a reccuring event. Risk is lower where foundation is mine fill. Candidate locations could be mitigated with additional erosion protection.				
W11	Ditch C	Thaw degradation leading to differential settlment of channel/inlet	Need for repairs	CC	Possible	Very Low	Low	NEW					
W12	Ditch C	Flows exceed Ditch C level of erosion protection resulting in damage to inlet structure/ditch and mobilization of tailings downstream	cleanup and Repair requirements	CC	Very Unlikely	Major	Moderate	NEW					
W13	Ditch C	Flows exceed Ditch C level of erosion protection resulting in damage to inlet structure/ditch and mobilization of tailings downstream	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	NEW					
W14	Ditch C	Failure of Main Pit north wall blocks inlet to Ditch C, leading to backing up of water and sudden release of water into Ditch C, damaging conveyance structure	Need for repairs	CC	Unlikely	Minor	Low	NEW					
W15	Ditch D	Thaw degradation leading to differential settlement of Channel in Ditch D	Need for repairs	CC	Very Unlikely	Very Low	Low	Y	large cut of Channel D lowers likelihood				
W16	Ditch D	Thaw degradation leading to differential settlement in Ditch D leading to ponding, overtopping, and erosion	Need for repairs	CC	Very Unlikely	Minor	Low	Y	large cut of Channel D lowers likelihood				
W17	Ditch D	Flows exceed Ditch D capacity resulting in failure of water conveyance structures	Repair requirements	CC	Very Unlikely	Very Low	Low	NEW					
W18	Ditch D	Channel side slope failure blocks inlet to Ditch D, leading to backing up of water in A2 Pit and sudden release of water into Ditch D, damaging conveyance structure	Need for repairs	CC	Very Unlikely	Minor	Low	NEW	Cut mostly in waste rock, potential for failure low				

TABLE B-4: Water Conveyance Structure Risk Register

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Risk
Low
Moderate
Moderately High
High
Very High

Water Conveyance (W)										Mitigated		
RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING L	RISK RATING C	CLASSIFICATION	changed?	COMMENTS	Potential Mitigations	RISK RATING L	RISK RATING C	CLASSIFICATION
W19	Ditch E	Flows exceed Ditch E capacity resulting in failure of water conveyance structures and erosion of MVFE cover, mbilization of sediments into CWTS head pond	Repair requirements for cover and ditch and cleanout of sediments in CWTS head pond	CC	Very Unlikely	Minor	Low	NEW					
W20	Ditch E	Thaw degradation leading to differential settlement in Ditch D leading to ponding, overtopping, and erosion	Need for repairs	CC	Very Unlikely	Minor	Low	NEW	potential for thaw consolidation in this area is less than that which would lead to overtopping of ponded water in Ditch E				
W21	Ditch E	Thaw degradation leading to differential settlement in Ditch E leading to ponding, overtopping and erosion resulting in sediment loading into passive treatment system	Unacceptable water quality conditions downstream of site	EI	Very Unlikely	Moderate	Low	N					
W22	CWTS High Flow Bypass	Landslide dam forms in CWTS high flow bypass channel as a result of thawing of permafrost overburden in S valley wall	Impounding of water and subsequent rapid breach and sediment loading downstream	EI	Possible	Minor	Moderate	Y	Included in post-closure monitoring - geotechnical inspections				
W23	CWTS High Flow Bypass	Landslide dam forms in CWTS high flow bypass channel as a result of thawing of permafrost overburden in S valley wall	Impounding of water and subsequent rapid breach and sediment loading downstream and need for repairs of conveyance and possibly slope	CC	Possible	Minor	Moderate	NEW					
W24	CWTS High Flow Bypass	Blockage forms in CWTS high flow bypass channel from woody debris (beaver/fallen trees)	Impounding of water and subsequent diversion into wetland causing damage to CWTS	CC	Possible	Minor	Moderate	NEW					
W25	Primary Channels	Inadequate design of primary network	Need for repairs	CC				Y	To broad to be useful, risks are addressed on specific feature basis.				
W26	Primary Channels	Flows exceed channel capacity resulting in failure of water conveyance structures	Need for repairs	CC				Y	To broad to be useful, risks are addressed on specific feature basis.				
W27	Secondary Channels	Excessive concentrated flows lead to erosion and gulley formation and mass wasting	Unacceptable sediment load downstream	EI	Unlikely	Minor	Low	Y	secondary channels are designed/armoured, erosion potential low				
W28	Secondary Channels	Excessive concentrated flows lead to erosion and gulley formation leading to waste rock exposure and infiltration	Unacceptable water quality downstream	EI	Unlikely	Moderate	Moderate	Y	changed to unlikely to reflect chance of moderate scale consequence, and moderate to be consistent with previous consequence rankings	advancements in design, progress on regrading at site	Very Unlikely	Moderate	Low
W29	Secondary Channels	Excessive concentrated flows lead to erosion and gulley formation	Repair requirements	CC	Unlikely	Very Low	Low	Y	Requires greater than design criteria flows to be any more than unlikely, sheet erosion analysis has been undertaken; site presence in PC will ensure site presence when these repairs are most likely required				
W30	Secondary Channels	Inadequate design of secondary network	Need for repairs	CC	Unlikely	Very Low	Low	Y	secondary network now designed for 1:200 yr event; site presence in PC will ensure site presence when these repairs are most likely required				
W31	Secondary Channels	Thaw consolidation near DSTSF leading to differential settlement under channel leading to ponding, overtopping, and erosion/tailings mobilization	Unacceptable water quality downstream	EI	Very Unlikely	Moderate	Low	NEW	Rating assumed erosion on South side where no rock shell exists				
W32	Secondary Channels	Thaw consolidation near DSTSF leading to differential settlement under channel leading to ponding, overtopping, and erosion/tailings mobilization	repairs required beyond regular maintenance	CC	Possible	Minor	Moderate	NEW	Rating assumed erosion on South side where no rock shell exists	refine landform design to reduce length of individual secondary channels and associated catchment areas	Very Unlikely	Minor	Low
W33	Tertiary Channels	Differential settlement leading to excessively concentrated flows in channels	scour and erosion and filling of the energy dissipator structures	CC	Unlikely	Minor	Low	Y	These networks are typically on flatter areas, which cannot generate velocities required for substantial erosion. Requires greater than design criteria flows to be any more than unlikely, sheet erosion analysis has been undertaken; site presence in PC will ensure site presence when these repairs are most likely required				

TABLE B-4: Water Conveyance Structure Risk Register

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Risk
Low
Moderate
Moderately High
High
Very High

Water Conveyance (W)											Mitigated		
RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING L	RISK RATING C	CLASSIFICATION	changed?	COMMENTS	Potential Mitigations	RISK RATING L	RISK RATING C	CLASSIFICATION
W34	Tertiary Channels	Differential settlement leading to excessively concentrated flows in channels leading to waste rock exposure and infiltration	Unacceptable water quality downstream	EI				Y	removed: see WRD rankings				
W35	Tertiary Channels	Differential settlement leading to excessively concentrated flows in channels leading to waste rock exposure and infiltration	Repair requirements	CC				Y	removed: see WRD rankings				
W36	Tertiary Channels	Inadequate design of tertiary network	Need for repairs	CC	Unlikely	Minor	Low	Y	have completed erosion analysis with has illustrated low sensitivity to design details. Adequate vegetation cover is most important measure to adequately control erosion. site presence in PC will ensure site presence when these repairs are most likely required				
W37	Tertiary Channels	Vegetation growth is less than expected leading to concentrated flows, erosion and sedimentation of downstream conveyance structures	need for cleanup/repairs	CC	Unlikely	Minor	Low	Y	Risk mitigated through existing plan for periodic maintenance and in Closure AMP. Landform design has been advanced				

TABLE B-5: Administrative Risk Register

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Administrative (A)									
RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING		CLASSIFICATION	changed?	COMMENTS
					L	C			
A1	Site Wide	General failure to conduct preventative mainenance and corrective actions leading to system failures (passive treatment, covers, etc.)	Impacts on ecosystem components	EI	Unlikely	Major	Moderately High	Y	Cost allowance now included for long term monitoring and maintenance activities, and facility design improvements have reduced exposure to risk.
A2	Site Wide	General failure to conduct preventative mainenance and corrective actions leading to system failures (passive treatment, covers, etc.)	* undefined *	CC				N	Not rated in 2014, revisited, still no definable cost associated with this.
A3	Site Wide	General failure to conduct preventative mainenance and corrective actions leading to system failures (passive treatment, covers, etc.)	H&S impacts	HHS	Unlikely	Minor	Low	N	
A4	Site Wide	Departure from design of engineered structures during construction	Unacceptable water quality conditions downstream	EI	Possible	Moderate	Moderately High	N	This ranking assumes violation of future licence conditions that authorize construction of engineered structures, and failure of regulatory oversight during/following construction. Will be mitigated with appropriate QA/QC monitoring, as per standard construction practice. 2017 Participants strongly disagree with this risk rating (from 2014 workshop) as it is based on future dicision-making, but it has not been changed as there are no conceivable current ways to mitigate this risk.
A5	Site Wide	Departure from design of engineered structures	Need for upgrades/ repairs/ redesign	CC	Possible	Major	High	N	This ranking assumes violation of future licence conditions that authorize construction of engineered structures, and failure of regulatory oversight during/following construction. Will be mitigated with appropriate QA/QC monitoring, as per standard construction practice. 2017 Participants strongly disagree with this risk rating (from 2014 workshop) as it is based on future decision-making, but it has not been changed as there are no conceivable current ways to mitigate this risk.
A6	Site Wide	Bankruptcy/ dissolution of the company and inadequate financial security	Requirement for public government to fund and conduct preventative maintenance and corrective actions to avoid system failures and impacts on ecosystem components	CC	Possible	Major	High	N	Likelihood rating was initially selected (2014) to reflect that this scenario has happened elsewhere. The 2017 participants strongly disagree with the 2014 risk ranking, as there has been a substantial increase in financial security. However, the risk ranking has not been changed.
A7	Site Wide	Failure of institutional controls	Land use that causes unanticipated negative exposure of humans or wildilife	HHS	Very Unlikely	Critical	Moderately High	N	Rated assuming failure results in a human fatality. Likelihood rating considers remoteness of site. Access to site in post closure remains a future land-use management decision requiring resolution. Risk ranking not changed in 2017.
A8	Site Wide	Failure of quality control during cover construction results in use of overburden materials that do not meet specifications	revegetation objectives not being achieved and additional work required to replace portions of cover	CC	Very Unlikely	Major	Moderate	NEW	ranking assumes example of full replacement of cover on DSTSF

TABLE B-6: Water Treatment Risk Register

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Water Treatment (WT)											Mitigated		
RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING		CLASSIFICATION	changed?	COMMENTS	Additional Potential Mitigations	RISK RATING		CLASSIFICATION
					L	C					L	C	
WT1	Source Terms	Waste rock upgradient of pits has source water quality worse than predicted, resulting in increased concentrations in pit lakes and in CWTS influent, and increased loadings in CWTS effluent	significant impact to downstream aquatic resources	EI	Unlikely	Moderate	Moderate	Y	assumes AMP is followed				
WT2	Source Terms	Waste rock upgradient of pits has source water quality worse than predicted, resulting in increased concentrations in pit lakes and in CWTS influent, and increased loadings in CWTS effluent	minor localized or short-term impacts to downstream aquatic resources	EI	Unlikely	Minor	Low	NEW	assumes AMP is followed				
WT3	Source Terms	Waste rock upgradient of pits has source water quality worse than predicted, resulting in increased concentrations in pit lakes and in CWTS influent, and increased loadings in CWTS effluent	resulting in implementation of mitigation measures	CC	Unlikely	Critical	Moderately High	N	assumes implementation of responses in AMP lead to long term active water treatment				
WT4	Source Terms	Pit walls have source water quality worse than predicted, resulting in increased concentrations in pit lakes and in CWTS influent, and increased loadings in CWTS effluent	significant impact to downstream aquatic resources	EI	Unlikely	Moderate	Moderate	NEW	assumes AMP is followed				
WT5	Source Terms	Pit walls have source water quality worse than predicted, resulting in increased concentrations in pit lakes and in CWTS influent, and increased loadings in CWTS effluent	minor localized or short-term impacts to downstream aquatic resources	EI	Unlikely	Minor	Low	NEW	assumes AMP is followed				
WT6	Source Terms	Pit walls have source water quality worse than predicted, resulting in increased concentrations in pit lakes and in CWTS influent, and increased loadings in CWTS effluent	resulting in implementation of mitigation measures	CC	Unlikely	Moderate	Moderate	NEW	assumes implementation of responses in AMP				
WT7	Source Terms	In-pit tailings has source water quality worse than predicted, resulting in increased concentrations in pit lakes and in CWTS influent, and increased loadings in CWTS effluent	significant impact to downstream aquatic resources	EI	Unlikely	Moderate	Moderate	NEW	assumes AMP is followed				
WT8	Source Terms	In-pit tailings has source water quality worse than predicted, resulting in increased concentrations in pit lakes and in CWTS influent, and increased loadings in CWTS effluent	minor localized or short-term impacts to downstream aquatic resources	EI	Unlikely	Minor	Low	NEW	assumes AMP is followed				
WT9	Source Terms	In-pit tailings has source water quality worse than predicted, resulting in increased concentrations in pit lakes and in CWTS influent, and increased loadings in CWTS effluent	resulting in implementation of mitigation measures	CC	Unlikely	Moderate	Moderate	NEW	assumes implementation of responses in AMP				
WT10	Source Terms	Dry Stack tailings has source water quality worse than predicted, resulting in increased concentrations in CWTS influent, and increased loadings in CWTS effluent	significant impact to downstream aquatic resources	EI	Unlikely	Moderate	Moderate	NEW	assumes AMP is followed				
WT11	Source Terms	Dry Stack tailings has source water quality worse than predicted, resulting in increased concentrations in CWTS influent, and increased loadings in CWTS effluent	minor localized or short-term impacts to downstream aquatic resources	EI	Possible	Minor	Moderate	NEW	assumes very short duration, i.e. failure mode is an ice blockage at the toe of the MFVE that impounds seepage and then melts in spring, resulting in a quick drainage of impounded water				
WT12	Source Terms	Dry Stack tailings has source water quality worse than predicted, resulting in increased concentrations in CWTS influent, and increased loadings in CWTS effluent	resulting in implementation of mitigation measures	CC	Unlikely	Critical	Moderately High	N	assumes implementation of responses in AMP lead to long term active water treatment				
WT13	Source Terms	MVFE has source water quality worse than predicted, resulting in increased concentrations in CWTS influent, and increased loadings in CWTS effluent	significant impact to downstream aquatic resources	EI	Unlikely	Moderate	Moderate	NEW	assumes AMP is followed				
WT14	Source Terms	MVFE has source water quality worse than predicted, resulting in increased concentrations in CWTS influent, and increased loadings in CWTS effluent	minor localized or short-term impacts to downstream aquatic resources	EI	Unlikely	Minor	Low	NEW	assumes AMP is followed				
WT15	Source Terms	MVFE has source water quality worse than predicted, resulting in increased concentrations in CWTS influent, and increased loadings in CWTS effluent	resulting in implementation of mitigation measures	CC	Unlikely	Moderate	Moderate	NEW	assumes implementation of responses in AMP				
WT16	Minto North Source Terms	Minto North pit walls have source water quality worse than predicted, resulting in increased loadings to McGinty Creek	significant impact to downstream aquatic resources	EI	Very Unlikely	Moderate	Low	NEW	no significant valued ecosystem components documented in McGinty Creek				
WT17	Minto North Source Terms	Minto North pit walls have source water quality worse than predicted, resulting in increased loadings to McGinty Creek	minor localized or short-term impacts to downstream aquatic resources	EI	Very Unlikely	Minor	Low	NEW	no significant valued ecosystem components documented in McGinty Creek				
WT18	Minto North Source Terms	Minto North pit walls have source water quality worse than predicted, resulting in increased loadings to McGinty Creek	resulting in implementation of mitigation measures	CC	Likely	Minor	Moderate	NEW	assumes implementation of responses in AMP				

TABLE B-6: Water Treatment Risk Register

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Water Treatment (WT)											Mitigated		
RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING L	RISK RATING C	CLASSIFICATION	changed?	COMMENTS	Additional Potential Mitigations	RISK RATING L	RISK RATING C	CLASSIFICATION
WT19	Water Treatment	Blockage of high flow bypass prior to a high flow event leads to routing of all high flows through CWTS, compromising CWTS cell structure, leading to short circuiting and/or compromising of CWTS vegetation and subsequent functionality at routine flows	significant impact to downstream aquatic resources	EI	Unlikely	Moderate	Moderate	NEW	Extensive damage and performance impairment is possible, but unlikely to result in significant impact to downstream resources				
WT20	Water Treatment	Blockage of high flow bypass prior to a high flow event leads to routing of all high flows through CWTS, flushing of sediments downstream, and release of sediment-bound metals to the aquatic environment	significant impact to downstream aquatic resources	EI	Very Unlikely	Moderate	Low	NEW	Likelihood ranking is based on low likelihood of damage (above) and low potential for sediment mobilization, as the location of sediment-bound metals is under protective vegetation.				
WT21	Water Treatment	Blockage of high flow bypass prior to a high flow event leads to routing of all high flows through CWTS, compromising CWTS cell structure, leading to short circuiting and/or compromising of CWTS vegetation and subsequent functionality at routine flows	requiring maintenance and replanting beyond regular planned maintenance	CC	Possible	Moderate	Moderately High	NEW	likelihood of extensive damage is possible				
WT22	Water Treatment	Sedimentation/erosion in cells leads to channelization and non-uniform flow field, reducing CWTS treatment efficacy	significant impact to downstream aquatic resources	EI	Unlikely	Moderate	Moderate	NEW					
WT23	Water Treatment	Sedimentation/erosion in cells leads to channelization and non-uniform flow field, reducing CWTS treatment efficacy	intermittent exceedances of water quality objectives but not effects thresholds	EI	Possible	Very Low	Low	NEW	Routine monitoring and maintenance would reduce likelihood of this happening to a degree that would cause water quality objective exceedences				
WT24	Water Treatment	Willows or alders grow into wetland area, leading to channeling caused by roots and reduced treatment retention time	significant impact to downstream aquatic resources	EI	Very Unlikely	Moderate	Low	NEW					
WT25	Water Treatment	Willows or alders grow into wetland area, leading to channeling caused by roots and reduced treatment retention time	requiring repair beyond regular planned maintenance	CC	Very Unlikely	Minor	Low	NEW	Routine monitoring and maintenance would reduce likelihood of this happening to a degree that would cause water quality objective exceedences				
WT26	Water Treatment	Willows or alders grow into wetland area, leading to increased vegetation biodiversity and attraction of wildlife and disturbance of treatment system	intermittent exceedances of water quality objectives but not effects thresholds	EI	Very Unlikely	Minor	Low	NEW					
WT27	Water Treatment	Willows or alders grow into wetland area, leading to increased vegetation biodiversity and attraction of wildlife and disturbance of treatment system	requiring repair beyond regular planned maintenance	CC	Very Unlikely	Minor	Low	NEW	Routine monitoring and maintenance would reduce likelihood of this happening to a degree that would cause water quality objective exceedences				
WT28	Water Treatment	Aphid infestation damages CWTS vegetation, decreasing carbon yield and compromising subsequent years' treatment performance	intermittent exceedances of water quality objectives but not effects thresholds	EI	Unlikely	Minor	Low	NEW	Routine monitoring and maintenance would reduce likelihood of this happening to a degree that would cause water quality objective exceedences				
WT29	Water Treatment	Aphid infestation damages CWTS vegetation, decreasing carbon yield and compromising subsequent years' treatment performance	requiring pest control/replanting beyond regular planned maintenance	CC	Possible	Minor	Moderate	NEW					
WT30	Water Treatment	CWTS vegetation growth is insufficient to supply required carbon to feed treatment process	intermittent exceedances of water quality objectives but not effects thresholds	EI	Unlikely	Minor	Low	NEW					
WT31	Water Treatment	CWTS vegetation growth is insufficient to supply required carbon to feed treatment process	requiring amendment beyond regular planned maintenance	CC	Unlikely	Minor	Low	NEW	Routine monitoring and maintenance would reduce likelihood of this happening to a degree that would cause need for amendment addition				
WT32	Water Treatment	Inflow to CWTS from head pond is blocked, leading to all flow entering high flow bypass	intermittent exceedances of water quality objectives but not effects thresholds	EI	Likely	Minor	Moderate	NEW					
WT33	Water Treatment	Inflow to CWTS from head pond is blocked, leading to all flow entering high flow bypass and drying out of CWTS, compromising system functionality while vegetation re-establishes	significant impact to downstream aquatic resources	EI	Very Unlikely	Moderate	Low	NEW					
WT34	Water Treatment	lack of replacement for CWTS intallation in plan/costing at end of CWTS design life	significant impact to downstream aquatic resources	EI	Unlikely	Moderate	Moderate	NEW	RCP does not include CWTS replacement, by design. Assumes site loadings have decreased to levels not requiring polishing.				

TABLE B-6: Water Treatment Risk Register

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Water Treatment (WT)											Mitigated		
RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING L	RISK RATING C	CLASSIFICATION	changed?	COMMENTS	Additional Potential Mitigations	RISK RATING L	RISK RATING C	CLASSIFICATION
WT35	Water Treatment	more than expected sedimentation in head pond and CWTS from upstream material sluffing into channels	requiring repair beyond regular planned maintenance	CC	Unlikely	Minor	Low	NEW	assumes an additional maintenance event is required				
WT36	Area 2 Pit	overturn of seasonal thermally stratified pit water, resulting in higher than expected pit discharge concentrations for short time periods	intermittent exceedances of water quality objectives but not effects thresholds	EI	Unlikely	Minor	Low	Y	Unlikely because of limited source loadings to pit				
WT37	Area 2 Pit	Discharge concentrations higher than expected when pit first discharges	short/medium term exceedances of water quality objectives but not effects thresholds	EI	Very Unlikely	Minor	Low	y	Pit water quality assumed to be carefully monitored during the transition stage; (pit takes ~ 3years to fill); Assumes treatment occurs if required.				
WT38	Area 2 Pit	Discharge concentrations higher than expected when pit first discharges	Requires treatment to meet acceptable discharge quality by time of first discharge	CC	Likely	Moderate	Moderately High	y	Pit water quality assumed to be carefully monitored during the transition stage; (pit takes ~ 3years to fill). Increased likelihood from 2014.				
WT39	Minto North Pit	Pit water quality degraded unacceptably	Unacceptable water quality conditions for water fowl / wildlife	EI	Very Unlikely	Moderate	Low	NEW	Scenario was evaluated				
WT40	Main Pit	overturn of seasonal thermally stratified pit water, resulting in higher than expected pit discharge concentrations for short time periods	intermittent exceedances of water quality objectives but not effects thresholds	EI	Unlikely	Minor	Low	NEW	shallower final depth in closure than A2 Pit, so less likely to stratify, however loading sources are higher in Main Pit, therefore rated same likelihood as A2 Pit.				
WT41	Main Pit	Discharge concentrations higher than expected when pit first discharges	short/medium term exceedances of water quality objectives but not effects thresholds	EI	Very Unlikely	Minor	Low	NEW	Pit water quality assumed to be carefully monitored during the transition stage; Assumes treatment occurs if required.				
WT42	Main Pit	Discharge water quality objectives are not met when pit first discharges	Requires treatment to meet discharge objectives by time of first discharge	CC	Likely	Moderate	Moderately High	NEW	Pit water quality assumed to be carefully monitored during the transition stage; Assumes treatment occurs if required.				
WT43	Main Pit	Wave action and suspension of tailings mobilizes tailings material out of MPTMF	intermittent exceedances of water quality objectives but not effects thresholds	EI	Possible	Minor	Moderate	NEW		will be mitigated by ensuring that tailings cover in MPTMF will extend to 1m depth below final	Very Unlikely	Minor	Low
WT44	Primary Channels	SWD toe seepage collection systems inadequate	Unacceptable WQ downstream	EI	Very Unlikely	Moderate	Low	Y	Reclamation research (seepage monitoring) has not identified substantial loading that could be mitigated by a seepage collection system				

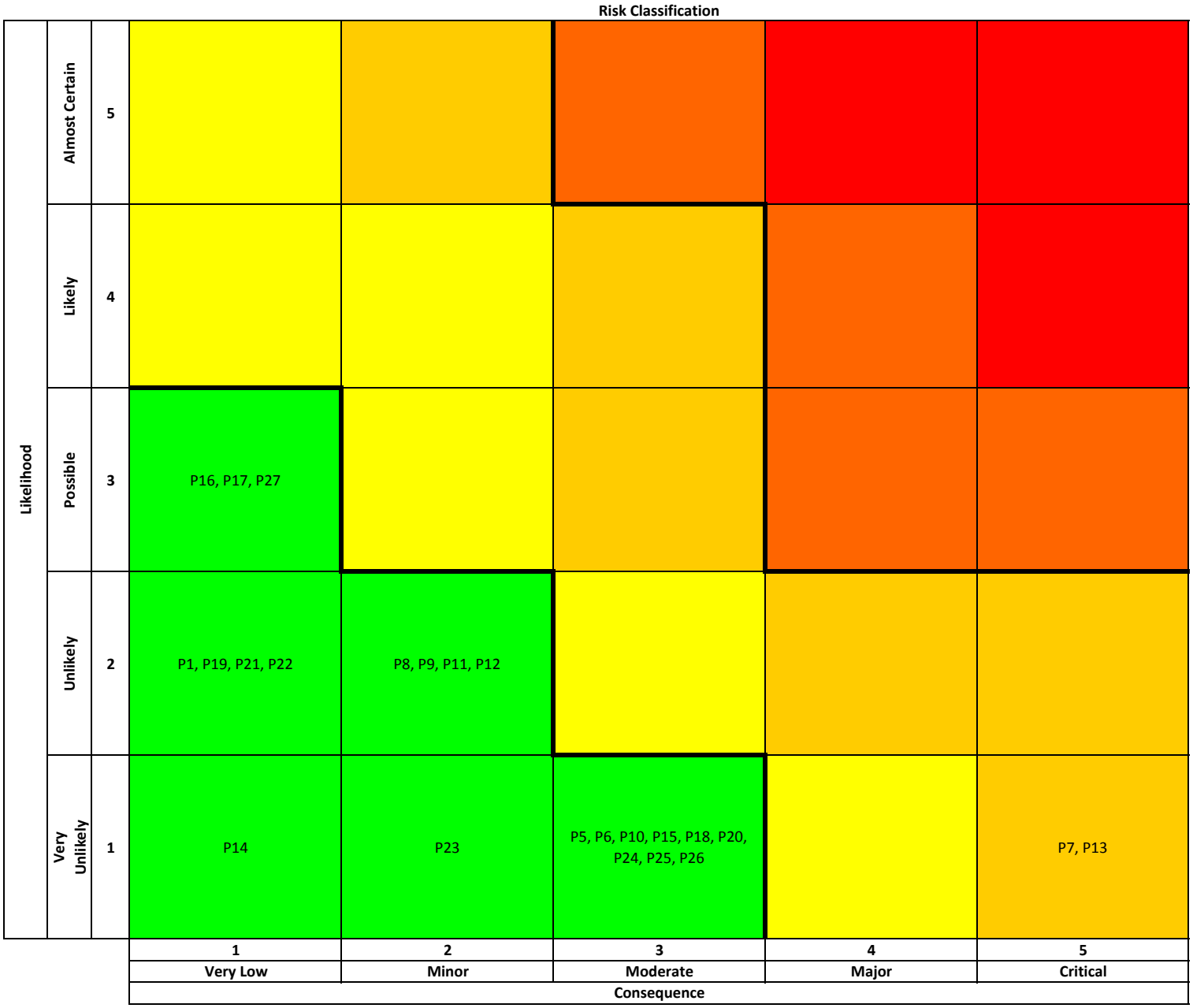
APPENDIX C.

2017 FMEA Risk Matrices

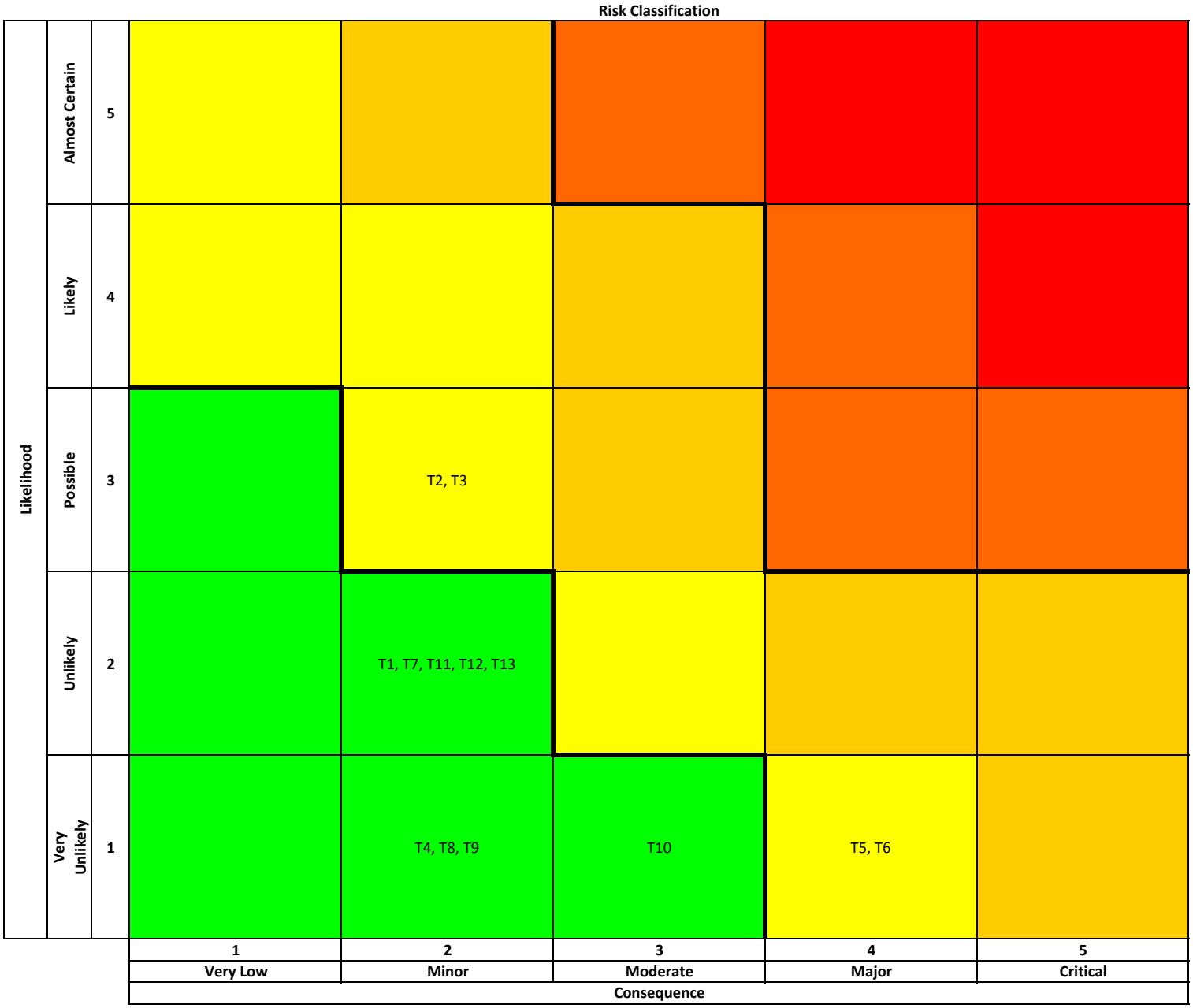
Risk Matrix - Overall Summary (2018-01)

			Risk Classification				
	Almost Certain	5					
	Likely	4		WT18, WT32, W10	WT38, WT42		
	Possible	3	P16, P17, P27, WT23, W11	T2, T3, D5, D12, D14, D18, D19, D29, D30, D30, WT11, WT29, W22, W23, W24	WT21, A4	A5, A6	
	Unlikely	2	P1, P19, P21, P22, D10, D24, W29, W30	P8, P9, P11, P12, T1, T7, T11, T12, T13, D4, D6, D11, D13, D20, D31, D31, WT2, WT5, WT8, WT14, WT28, WT30, WT31, WT35, WT36, WT40, W5, W6, W14, W27, W33, W36, W37, A3	WT1, WT4, WT6, WT7, WT9, WT10, WT13, WT15, WT19, WT22, WT34	A1	WT3, WT12
	Very Unlikely	1	P14, D27, W3, W4, W15, W17	P23, T4, T8, T9, D23, D26, WT17, WT25, WT26, WT27, WT37, WT41, WT43, W1, W2, W7, W8, W16, W18, W19, W20, W32	P5, P6, P10, P15, P18, P20, P24, P25, P26, T10, D1, D2, D3, D7, D8, D9, D15, D16, D17, D21, D22, D25, WT16, WT20, WT24, WT33, WT39, WT44, W9, W13, W21, W28, W31	T5, T6, D28, W12, A8	P7, P13, A7
			1	2	3	4	5
			Very Low	Minor	Moderate	Major	Critical
			Consequence				

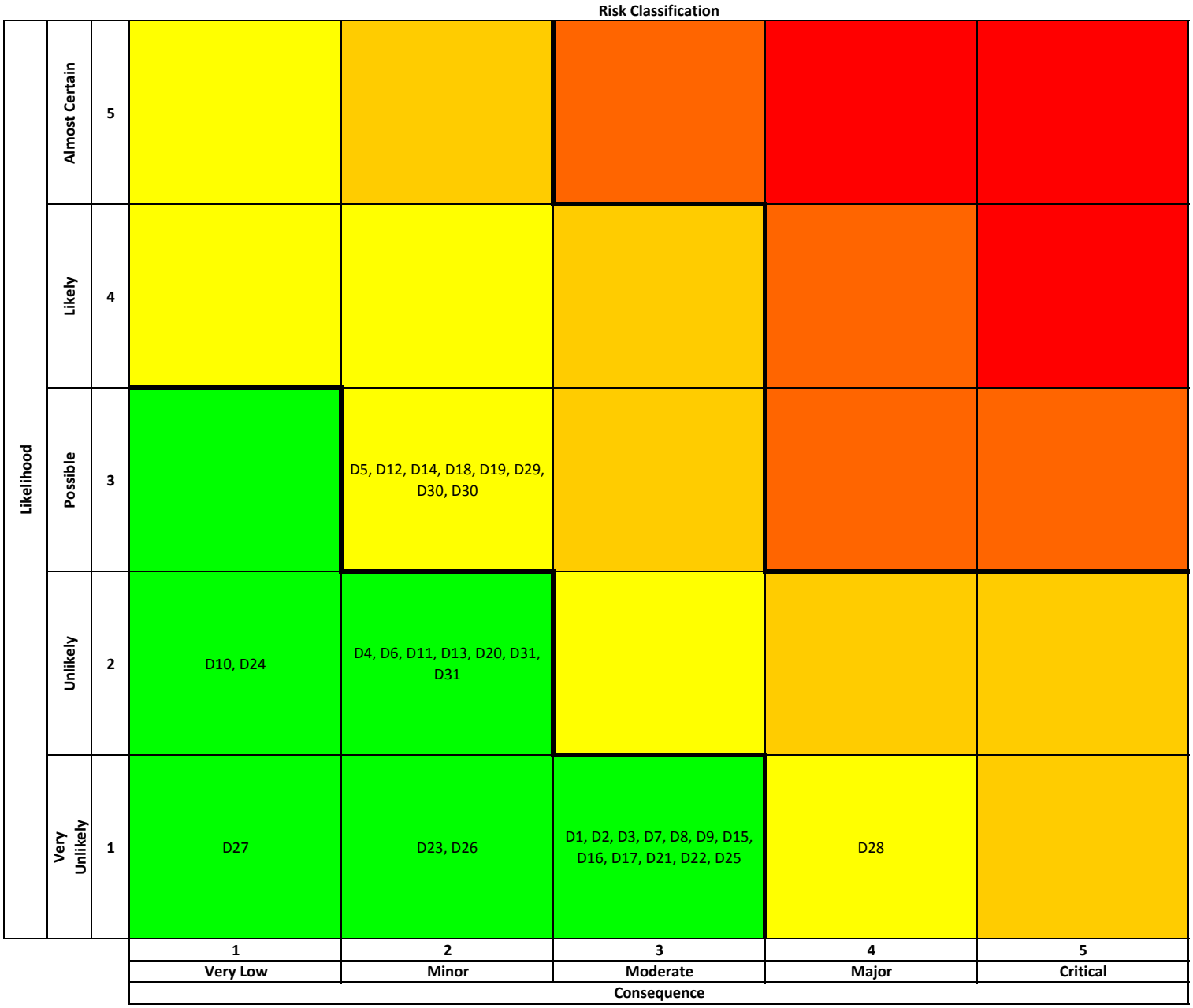
Risk Matrix - Open Pits (P) (2018-01)



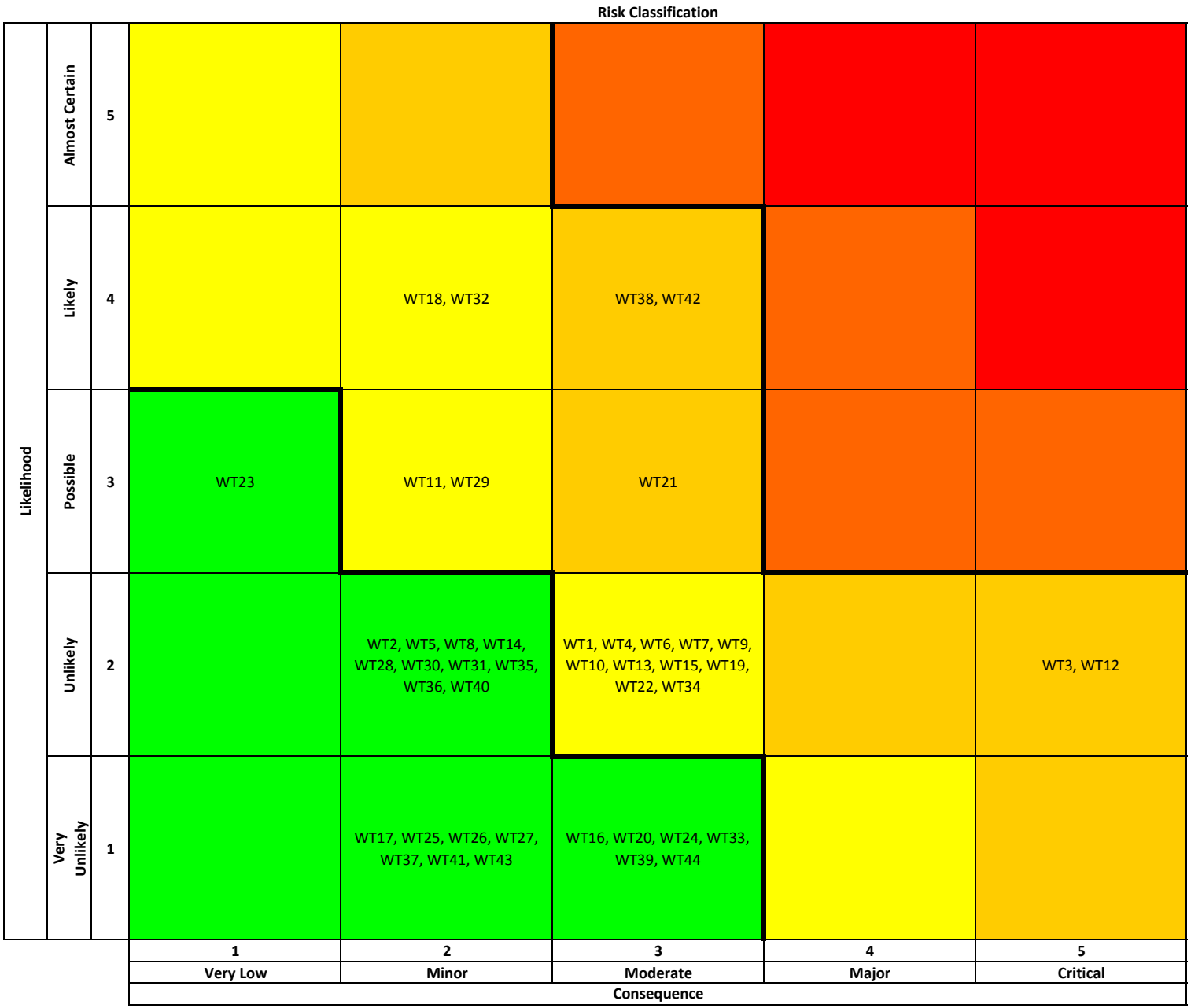
Risk Matrix - Dry Stack Tailings Storage Facility (T) (2018-01)



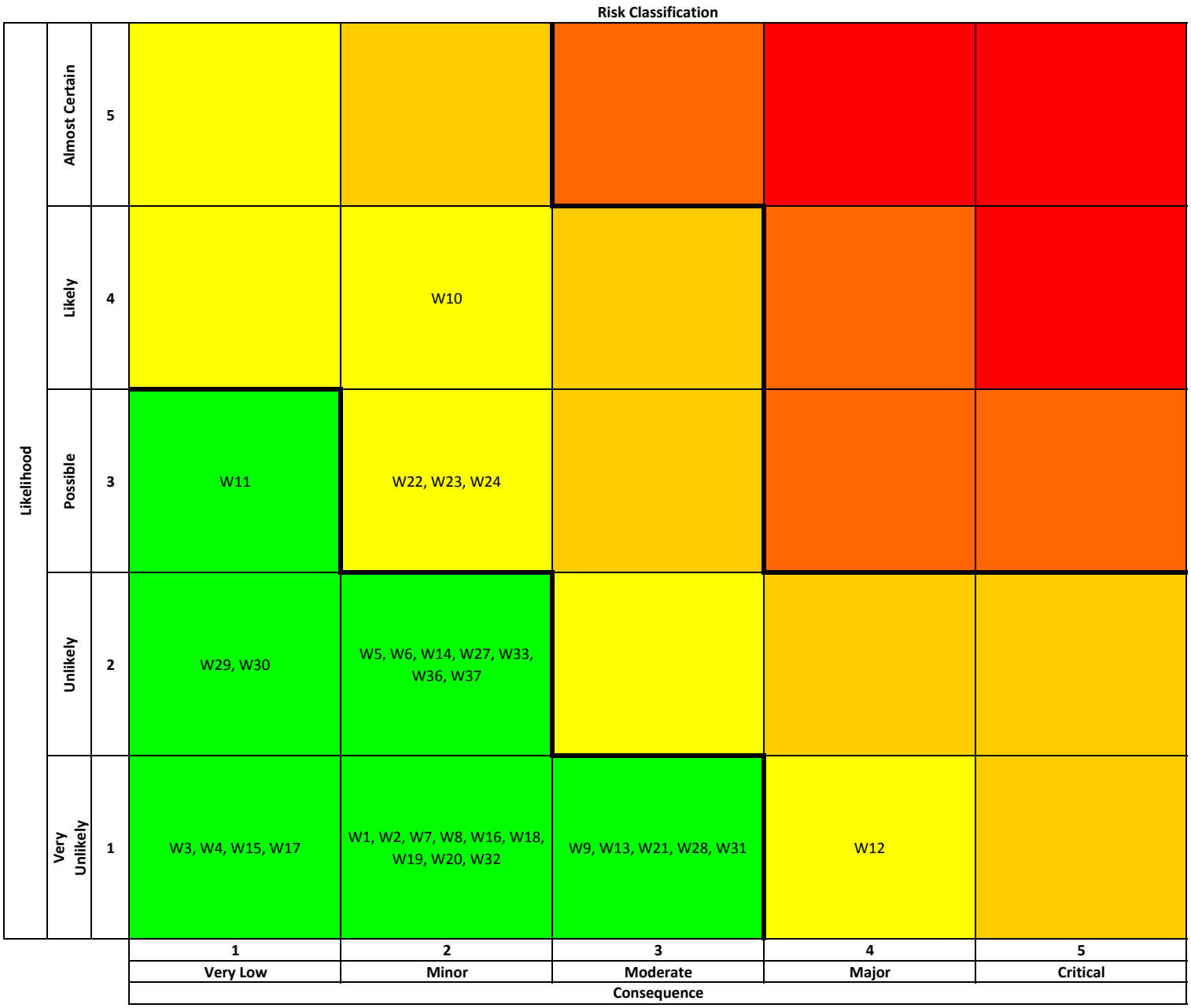
Risk Matrix - Waste Rock Dumps (D) (2018-01)



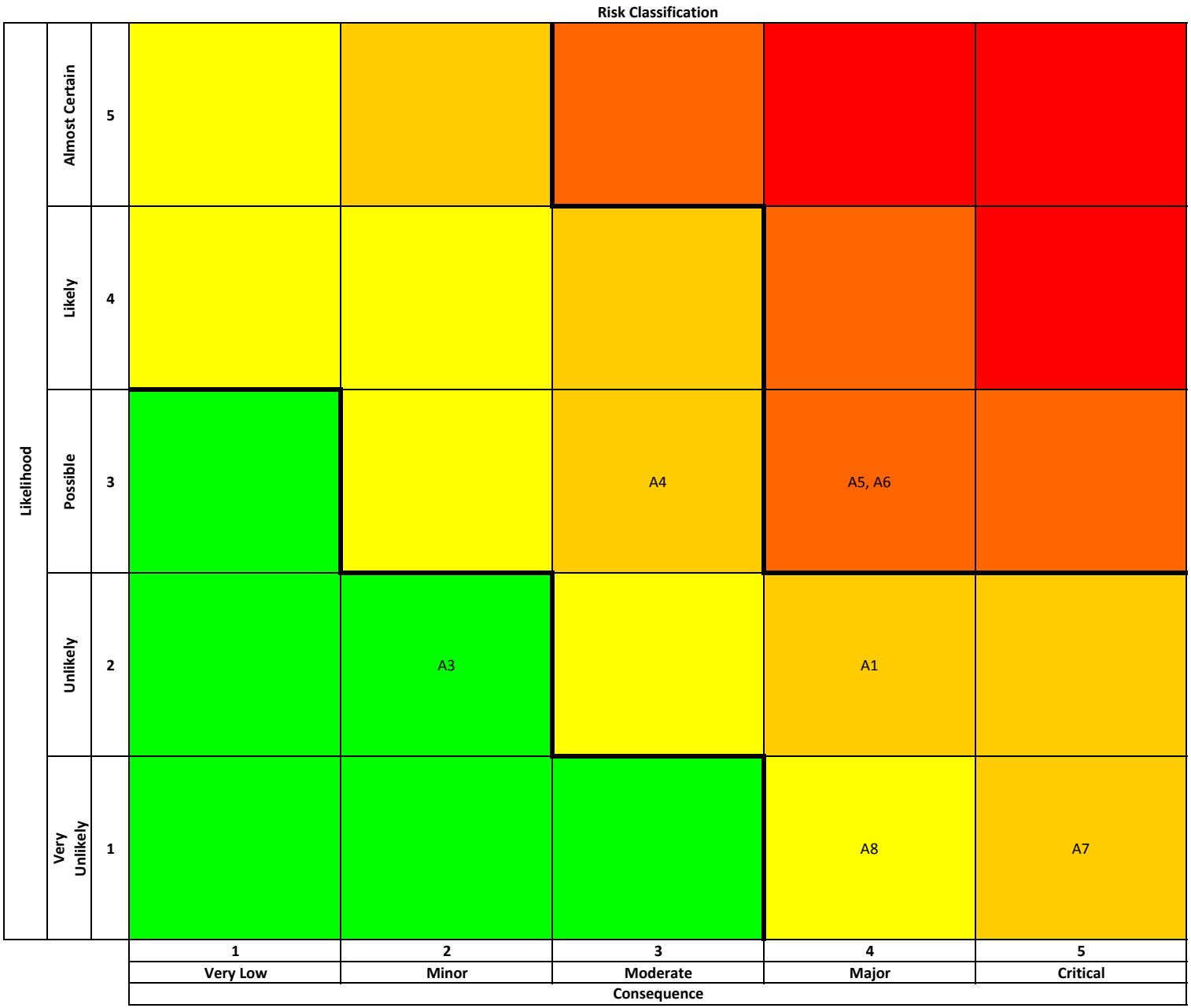
Risk Matrix - Water Treatment (WT) (2018-01)



Risk Matrix - Water Conveyance (W) (2018-01)



Risk Matrix - Administrative (A) (2018-01)



APPENDIX D.

Summary of Risks Not Re-Evaluated in 2017 FMEA

TABLE D-1: 2014 FMEA Entries not Re-evaluated

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Not Re-evaluated (NR)

RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING		CLASSIFICATION	2014 COMMENTS	Rationale for Not Re-evaluating Risk
					L	C			
R1	Area 2 Pit	Precipitation higher than design asssumption (1:200 yr 24 hr)	Erosion of Ditch 400 channel and damage to toe of Main Dam, leading to breach of dam and release of tailings to lower Minto Creek	TU	Very Unlikely	Major	Moderate		Consequence Category not utilized in 2017 Workshop
R2	Area 2 Pit	Precipitation higher than design asssumption (1:200 yr 24 hr)	Erosion of Ditch 400 channel and damage to toe of Main Dam, leading to breach of dam and release of tailings to lower Minto Creek	RL	Very Unlikely	Major	Moderate		Consequence Category not utilized in 2017 Workshop
R3	Area 2 Pit	Precipitation higher than design asssumption (1:200 yr 24 hr)	Erosion of Ditch 400 channel and damage to toe of Main Dam, leading to breach of dam and release of tailings to lower Minto Creek	CMR	Very Unlikely	Critical	Moderately High		Consequence Category not utilized in 2017 Workshop
R4	Area 2 Pit	Pit water quality degraded unacceptably (dissolved constituents)	Unacceptable water quality conditions for water fowl / wildlife	TU	Unlikely	Moderate	Moderate	Scenario needs to be evaluated; water quality that could affect water fowl needs to be researched and shared.	Consequence Category not utilized in 2017 Workshop
R5	Area 2 Pit	Pit water quality degraded unacceptably (dissolved constituents)	Problematic exceedances of site water quality discharge standards	RL	Possible	Moderate	Moderately High		Consequence Category not utilized in 2017 Workshop
R6	Minto North Pit	Positive water balance for pit leads to development of pit lake	results in negative perception leading to negative impacts on traditional land use	TU	Possible	Minor	Moderate	Could be mitigated through education and information sharing. Discussion by group noted that mitigation of perception is complicated by variability in perceptions among individuals.	Consequence Category not utilized in 2017 Workshop
R7	Minto North Pit	Pit water quality degraded unacceptably	Unacceptable water quality conditions for water fowl / wildlife and results in some mitigatable impact to tradtional land use	TU	Very Unlikely	Moderate	Low	Assumes AMP does not exist or is not implemented	Consequence Category not utilized in 2017 Workshop
R8	DSTSF/MVFE	Landslide dam forms in footprint of Water Storage Pond as a result of thawing of permafrost overburden in S valley wall due to presence of Water Storage Pond	Impounding of water and subsequent rapid breach and sediment loading downstream	EI	Possible	Minor	Moderate	Consider including in post-closure monitoring	Expanded upon in Water Convenyance Category
R9	Main Waste Dump	Precipitation higher than expected	Failure of water conveyance structures because structures are underdesigned					Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA	Addressed in Water Conveyance Category
R10	Main Waste Dump	Undiverted runoff upstream of waste mgmt facilities leads to runon water, extra infiltration	Ongoing maintenance costs					Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA	Addressed in Waste Rock Dump Category
R11	Southwest Waste Dump	SWD toe seepage collection systems inadequate	Unacceptable WQ downstream					Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA	Addressed in Water Treatment Category
R12	Southwest Waste Dump	Precipitation higher than expected	Failure of water conveyance structures because structures are underdesigned					Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA	Expanded upon in Water Convenyance Category

TABLE D-1: 2014 FMEA Entries not Re-evaluated

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Not Re-evaluated (NR)

RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING		CLASSIFICATION	2014 COMMENTS	Rationale for Not Re-evaluating Risk
					L	C			
R13	Southwest Waste Dump	Undiverted runoff upstream of waste mgmt facilities leads to runon water, extra infiltration	Ongoing maintenance costs					Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA	Addressed in Waste Rock Dump Category
R14	Southwest Waste Dump	Existing pond north of MGW/ south of IROD remains in post-closure	Causes community concern	CMR	Almost Certain	Very Low	Moderate	Will be revisited in detailed design and could be mitigated through education and information sharing	Consequence Category not utilized in 2017 Workshop
R15	Ridgetop Waste Dump	Precipitation higher than expected	Failure of water conveyance structures because structures are underdesigned					Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA	Addressed in Water Conveyance Category
R16	Ridgetop Waste Dump	Ridgetop Waste Dump name	Creates perception that there will be major viewshed impacts	CMR	Possible	Minor	Moderate	Could be mitigated through education and information sharing	Consequence Category not utilized in 2017 Workshop
R17	Reclamation OVB, Ridgetop South & Area 118 Backfill Dumps	Precipitation higher than expected	Failure of water conveyance structures because structures are underdesigned					Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA	Addressed in Waste Rock Dump Category
R18	Reclamation OVB, Ridgetop South & Area 118 Backfill Dumps	Undiverted runoff upstream of waste mgmt facilities leads to runon water, extra infiltration	Ongoing maintenance costs					Scenario wording copied from 2013 risk register-topic covered in #12 in 2014 FMEA	Addressed in Waste Rock Dump Category
R19	Reclamation OVB, Ridgetop South & Area 118 Backfill Dumps	Ponding of water on surface of overburden dumps leads to excessive infiltration, increases flow subsurface and contaminant loading from overburden	Unacceptable downstream water quality	EI	Very Unlikely	Moderate	Low	Designs of A118 and RS BD are mounded to shed water.	Addressed in Waste Rock Dump Category
R20	Reclamation OVB, Ridgetop South & Area 118 Backfill Dumps	Erosion on steeper portions of dumps	Sedimentation in conveyance channels and need for maintenance					Scenario topic covered in #12 in 2014 FMEA	Addressed in Waste Rock Dump Category
R21	Reclamation OVB, Ridgetop South & Area 118 Backfill Dumps	Storage of overburden in pits	Creates perception that valuable reclamation materials are being wasted	CMR	Almost Certain	Minor	Moderately High	Could be mitigated through education and information sharing	Consequence Category not utilized in 2017 Workshop
R22	Site wide	Source water quality (source term) worse than predicted	Problematic exceedances of downstream water quality objectives	CMR	Unlikely	Critical	Moderately High		Consequence Category not utilized in 2017 Workshop
R23	Main Dam	Thaw degradation leading to differential settlment of spillway, distruption of the armour layer, and scour of the spillway	Need for repairs	CC	Possible	Moderate	Moderately High	May be a recuring event. Configuation of spillway at closure needs to be considered. Consider emergency spillway during operations and different long term spillway. (consider in rock on north side). Likelyhood could be mitigated with routine maintenance.	Main Dam no longer part of Mine Plan/RCP

TABLE D-1: 2014 FMEA Entries not Re-evaluated

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Not Re-evaluated (NR)

RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING		CLASSIFICATION	2014 COMMENTS	Rationale for Not Re-evaluating Risk
					L	C			
R24	Main Dam	Thaw degradation leads to retrogressive failure at outlet of main pit spillway	Need for repairs	CC	Possible	Moderate	Moderately High	May be a recuring event. Configuation of spillway at closure needs to be considered. Consider emergency spillway during operations and different long term spillway. (consider in rock on north side). Likelyhood could be mitigated with routine maintenance.	Main Dam no longer part of Mine Plan/RCP
R25	Main Pit	Geotechnical failure of any waste facility (slope stability) resulting in debris dam, breaching, mobilizing materials and pulse of water into Main Pit, and sediments/tailings leaving pit	Unacceptable water quality conditions downstream of site	EI	Very Unlikely	Very Low	Low		Configuration of Main Pit at closure has changed - addressed in Pit Category
R26	Primary Channels	Increased infiltration through unlined Ditch 200 and Ditch 300	Unacceptable water quality conditions downstream of site	EI	Very Unlikely	Moderate	Low		Ditch configuration/design has changed - addressed in Water Conveyance Category
R27	Site wide (Admin)	General failure to conduct preventative mainenance and corrective actions leading to system failures (passive treatment, covers, etc.)	Negative traditional use	TU	Likely	Critical	Very High	-Assumes the existing traditional land use impacted due to perception; cease in trapping, hunting, berry gathering activity in area. -Assumes intended post-mining land use same as pre-mining land use	Consequence Category not utilized in 2017 Workshop
R28	Site wide (Admin)	General failure to conduct preventative mainenance and corrective actions leading to system failures (passive treatment, covers, etc.)	Regulatory/legal action	RL	Possible	Critical	High	-Assumes company remains the responsible company.	Consequence Category not utilized in 2017 Workshop
R29	Site wide (Admin)	General failure to conduct preventative mainenance and corrective actions leading to system failures (passive treatment, covers, etc.)	Community/media/reputation impacts	CMR	Likely	Critical	Very High	media and reputation aspects of the concequence severity description were disregarded in rating.	Consequence Category not utilized in 2017 Workshop
R30	Site wide (Admin)	Revegetation does not meet closure objectives relating to end land use	Negative traditional use	TU	Possible	Moderate	Moderately High	Can be mitigated through appropriate selection of end land use, determination of end land use goals, appropriate development of closure objectives to support those goals, and appropriate selection of veg species and revegetation methods	Consequence Category not utilized in 2017 Workshop
R31	Site Wide	Failure to adequately meet reporting requirements	Noncompliance	RL	Possible	Minor	Moderate		Consequence Category not utilized in 2017 Workshop

TABLE D-1: 2014 FMEA Entries not Re-evaluated

Minto Explorations Ltd.
Minto Mine
FAILURES MODES AND EFFECTS ASSESSMENT

Not Re-evaluated (NR)									
RISK NO.	AREA	RISK DESCRIPTION	CONSEQUENCE	CON. TYPE	RISK RATING		CLASSIFICATION	2014 COMMENTS	Rationale for Not Re-evaluating Risk
					L	C			
R32	Area 2 Pit	Pit water quality degraded unacceptably (dissolved constituents)	Problematic exceedances of downstream water quality objectives	EI	Possible	Moderate	Moderately High	Assumes AMP in place, and discussed it being a short term impact - particularly with current AMP which addresses site water and pit water quality, but retained it as a moderate severity.	Expanded upon in Pit Category
R33	Minto North Pit	Pit water quality degraded unacceptably	Problematic exceedances of downstream water quality objectives	EI	Possible	Moderate	Moderately High	Assumes AMP does not exist or is not implemented	Expanded upon in Pit Category

Scoping Level Cover Assessment for Minto Closure Covers

Prepared for

Minto Explorations Ltd.



Prepared by

SRK Consulting (Canada) Inc.
1CM002.007
August 2013

Scoping Level Cover Assessment for Minto Closure Covers

August 2013

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File Name: Minto_Final_Scoping Level Cover Assessment_1CM002 007_20130822_EMR_DBM_CCS

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srk consulting

Executive Summary

Minto Explorations Ltd. (Minto) is considering the use of covers as one of the remedial strategies for mine waste at their Minto Mine site in Yukon Territory, Canada. Mine waste at Minto site includes both waste rock and tailings (dry stack and in-pit tailings). At the end of Phase IV (the current phase of mining), there will be six distinct waste rock and overburden piles on site, and these will be expanded as future mine development takes place.

Geochemical characterization of the mine waste has confirmed that acid generation potential is low; however, neutral metal leaching will be a long-term concern. This scoping level cover assessment provides a comprehensive evaluation of appropriate cover design concepts that should be considered for the Minto site, taking into account site specific conditions and overall site wide closure objectives.

Mine waste covers are one remedial technology that can be used to manage drainage. Successful closure covers over mine waste facilities depend on many factors; however, two factors dominate decisions on what may be appropriate at any specific site, i.e., climatic conditions and locally available material (i.e., soils).

The evaluation demonstrates that the hydrologic and climatic regime at the Minto site is best suited to the use of water covers or infiltration reducing covers (i.e., barrier covers). In contrast, store-and-release and thermal covers are not likely to be successful. The available overburden soils at Minto are however not well suited for construction of low or very low infiltration barrier covers.

These soils are best used as isolation and vegetation supporting covers, and are expected to result in overall infiltration typically between 10 and 20% of mean annual precipitation. Periodic higher breakthrough events will occur. Soil amendments or use of synthetic products are therefore required to construct low to very low infiltration covers at the site. Conceptual designs of each of these different cover variants are proposed.

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1 Introduction

1.1 Background

Minto Explorations Ltd. (Minto) is considering the use of covers as a remedial strategy at their Minto Mine site in Yukon Territory, Canada. The general concepts of this plan are documented in Minto's current Decommissioning and Reclamation Plan (D&RP) (Minto 2011). A rigorous assessment of the proposed cover concepts taking into account site specific conditions has not been carried out and therefore the cover design concepts are not optimized. This work is intended to provide a rationale for how covers can best be utilized at Minto.

Mine waste at Minto includes both waste rock and tailings. At the end of Phase IV (the current phase of mining) there will be six distinct waste rock and overburden piles on site as illustrated on Figure 1, and these will be expanded as future mine development takes place.

Historic tailings deposition consisted of dry-stacked compacted filtered tailings. This deposition strategy was discontinued in October 2012, in favour of conventional low solids content (50 to 60% by mass) slurry tailings deposited sub-aqueously into Main Pit. Future tailings deposition will continue as hydraulically deposited tailings in Main and Area 2 Pits.

Geochemical characterization carried out on both waste rock and tailings confirms that, for the most part, the mine waste is not highly reactive, and has very low potential to generate acid; however, there is potential for neutral metal leaching (SRK 2013a). Mitigation strategies are therefore required to manage this leachate, including water treatment, waste relocation and waste encapsulation (i.e., closure covers).

1.2 Scope of Work

While it is understood that covers are one of many possible mitigation strategies that can be used to achieve site wide closure objectives, the scope of this study was to specifically determine what cover concepts are most likely to be successful at Minto, and how best to implement them. This was done by carefully evaluating site specific conditions that affect cover design, construction and ultimately long-term performance, including but not limited to design life, climate, material availability, waste characterization, seismicity, slope stability, erosion resistance, etc.

Once the cover concepts that are most likely to be successful at Minto were identified, appropriate cover functions were defined to specifically ensure overall site wide closure objectives would be met. Finally conceptual cover designs are presented that demonstrate actual implementation plans for all of the concepts.

1.3 Report Layout

Specific design elements that need to be considered in cover design are described in Section 2. Section 3 of this report describes the different cover types and whether they should be given consideration at Minto. Section 4 provides a generalized overview of the currently stated site wide closure objectives, complete with preliminary suggestions for cover functionality. Finally, different conceptual cover designs suitable for application at Minto, complete with expected performance criteria, are presented.

2 General Cover Design Elements

2.1 Cover Design Life

SRK does not believe that engineered soil covers can be expected to continue to perform in accordance with their original design intent for an infinite lifetime after construction (i.e., in perpetuity), especially if there is no monitoring and maintenance plan in effect. Therefore, it is appropriate to design soil covers and associated monitoring and maintenance plans with a finite lifespan in mind. The length of that time period should dictate to what standard the cover must be designed and constructed, and also what level of monitoring and maintenance would be required.

There are no specific guidelines or standards in Canada that specify a lifespan for covers, leaving the design decision to the discretion of the mine operator. Internationally, there are also no fixed criteria for cover lifespan; however, some mining companies explicitly adopt life spans of 100, 200 or even 500 year periods during which the cover is expected to perform in accordance with the original design intent (Rykaart *et al*. 2006).

The design life of a cover is complicated by the fact that some cover failure modes require inherently different approaches. For example, if Minto was to adopt a design life for the conceptual cover designs presented in this report of say 200 years, then it would be assumed that these covers would continue to perform the functions listed in Section 4 for this lifespan. However, portions of the cover design may have different design criteria, for example surface water conveyance channels may be designed to withstand 1:100 year storm events with an understanding that the necessary maintenance and repair would be carried out for the 200 year design life of the cover.

It would be premature for SRK to recommend a design life for the Minto covers in isolation of the overall closure design, as well as input from stakeholders. A risk assessment approach can be used during detailed design to evaluate the consequences of the different cover failure modes and what the optimal cover design life should be.

2.2 Site Climate

Site specific climate is one of the primary drivers that define what the most appropriate and sustainable cover may be for any given project area. The three dominating climatic parameters include the climatic water balance, air temperature, and radiation energy.

Site specific climate data is available with records from two onsite weather stations. Both stations have data from 2005 to 2011; however, the datasets are incomplete and generally too short to be useable or representative for a scoping level cover assessment. Regional climate data is available from the Pelly Ranch station, managed by Environment Canada. This station is located about 25 km north of Minto and has data dating back to 1955.

Based on the Pelly Ranch data, the mean annual total precipitation (MAP) for Minto is estimated to be about 335 mm. Roughly 50% of this precipitation falls as rain, with the remainder being the snow water equivalent of the annual snowpack. The maximum annual precipitation on record is 466 mm.

Monthly lake evaporation (aka potential evaporation) has been recorded at the Pelly Ranch station from 1965 to 2005 and the mean annual lake evaporation is 452 mm. Site specific data suggest that the mean annual evaporation is closer to 430 mm (Clearwater 2008) and 438 mm (Clearwater 2010). EBA (2010) applied an elevation correction to the Pelly Ranch data (1971-1990) to reduce the mean annual evaporation to 400 mm. SRK calculated the mean annual evaporation from first principles using the FAO method (FAO 1998) which computes to 410 mm evaporation per year. For the purpose of this study, the mean annual evaporation was selected based on the elevation correction presented by EBA (2010) to the Pelly Ranch evaporation data to yield a mean annual evaporation of 400 mm. The mean monthly climatic water balance based on this data is presented in Figure 2 and it demonstrates that between March and September, the climatic water balance is net negative, i.e., evaporation exceeds precipitation. This highly seasonal water balance deficit on a near neutral annual water balance makes it challenging to determine the most suitable cover type and a more in depth analysis of the climate is required.

Using the elevation corrected annual lake evaporation and precipitation data from Pelly Ranch from 1965 to 1995, the Thornthwaite and Mather (1955) Moisture Index was calculated to provide an indication of the site climate classification taking into consideration both water and temperature influences. This is presented in Figure 3 and confirms, generally, the site can be classified as Dry Subhumid; however, over the years the site spans both the Moist Subhumid and Semi-Arid climate zones.

The climatic water balance and climate classification are good indicators for what the most appropriate cover type would be at any given site and this is illustrated for Minto on Figure 4. This graph has been adapted from the Holdridge *et al*. (1971) life zone classification, as presented in the Global Acid Rock Drainage (GARD) Guide (INAP 2012). The data confirms that the site is best suited towards either water covers or infiltration controlling (i.e., barrier type) covers. Store-and-release and thermal covers on the other hand are not recommended for the site based on climatic conditions.

The mean monthly ambient air temperature for the site is also plotted on Figure 2. The annual average ambient air temperature is -4^{o}C. This leads to seasonal ground freezing, and site specific measurements suggest that the annual depth of frost penetration is less than 4 m. The site is also located in the discontinuous permafrost region of Canada and where permafrost is present, it is generally relative warm at about -1^{o}C.

Empirical methods were used to determine the annual depth of frost penetration at Minto, which amounts to between 2 and 3.5 m depending on soil porosity, water content and bulk density. Therefore, any low permeability barrier cover that could be damaged through freeze-thaw action would have to be constructed with a protection cover of at least 3 m.

Freezing temperatures also lead to a unique moisture distribution profile which is important when considering possible covers. During spring, significant surface runoff is present in the form of snowmelt; however, during this time the ground is still frozen and therefore water is less likely to infiltrate. When considering the climatic water balance illustrated in Figure 2, it appears that during spring (freshet) and early summer, when the most runoff is expected, the highest evaporation potential exists.

Figure 5 provides an overall climate wheel for the Minto site which summarizes key dates and timelines which will affect cover performance from both a hydrologic and physical perspective.

2.3 Potential Cover Materials

Locally available candidate cover soils consist of the overburden soils which can generally be classified as silty sands. Numerous geotechnical characterization programs have been carried out over the years including a dedicated cover soil characterization program in 2012 (SRK 2013b).

Three primary candidate cover soil sources have been identified as illustrated in Figure 1; (1) overburden stockpiled on the Main Waste Dump (MWD); (2) the Reclamation Overburden Dump (ROD); and (3) the Ice-Rich Reclamation Overburden Dump (IROD). Additional material is currently being developed from Area 2, Stage 2 Pit development and is being stockpiled on the Dry Stack Tailings Storage Facility and the Reclamation Overburden Dump. In total, it is estimated that there is about 2.8 Mt of these soils available for use as cover material.

Indicator property characterization testing has been completed on these soils (i.e., particle size distribution and Atterberg Limits) and the summarised results are presented in Figures 6 and 7. Their results demonstrate that there is minimal variability in the available material when considering its use a candidate cover source. Based on the overall grain size distribution envelope for all of the candidate cover soils (Figure 6), nine curves were initially selected to represent the possible range in cover performance (Figure 8).

Hydraulic testing (i.e., porosity, saturated hydraulic conductivity and Soil Water Characteristic Curves (SWCC)) was carried out on select samples and the results are summarized in Table 1 and Figure 9. Prior to the completion of the SWCC testing, six of the nine samples in Figure 8 were chosen and SWCCs were estimated using empirical methods. These resultant curves are illustrated in Figure 10. Comparisons of these estimated curves with the measured curves are presented in Figures 11 through 13. The final six calibrated SWCCs that were used in the cover analysis presented in this report are presented in Figure 14.

The MWD overburden material is sandy, lean clay (CL, in accordance with the Unified Soil Classification System). The porosity is estimated at about 36% and saturated hydraulic conductivity is about 1.38 x 10^{-7} m/sec. The air entry value is estimated to be between 23.1 and 71.9 kPa.

Materials from the ROD and IROD range from silty sands to clayey sands (SM-SC). Porosity is estimated at 29% to 31%, and saturated hydraulic conductivity ranges between 7.72 x 10^{-8} and 9.24 x 10^{-8} m/sec. The air entry value is estimated to range between 1.5 and 12.1 kPa.

Table 1: Hydraulic properties of overburden material from MWD, ROD, and IROD

Material Type	Porosity	Saturated Hydraulic Conducticity (m/s)	Field Capacity (VWC)	Wilting Point (VWC)	Storage Capacity (VWC)	Air Entry Value (kPa)
MWD1	36.0%	1.38 x 10^{-7}	0.339	0.223	0.116	71.9
MWD2	36.0%	1.38 x 10^{-7}	0.304	0.192	0.112	23.1
ROD2	31.0%	9.16 x 10^{-8}	0.260	0.160	0.100	12.1
ROD3	29.0%	7.72 x 10^{-8}	0.176	0.112	0.065	1.5
IROD1	31.0%	9.24 x 10^{-8}	0.243	0.127	0.116	4.8
IROD2	30.5%	8.79 x 10^{-8}	0.202	0.122	0.080	1.7

Three properties dominate the suitability of material as it relates to its use as a candidate cover material. First, for barrier covers intended to reduce infiltration, the saturated hydraulic conductivity is the most important. As a general rule of thumb, a soil should have a saturated hydraulic conductivity of at least 1 x 10^{-8} m/s (315 mm/year) to be considered an infiltration barrier.

Secondly, the property that dominates the suitability of a store-and-release type cover is how well graded it is, which translates into its moisture holding capacity (i.e., difference between its air entry value and residual suction). The moisture holding capacity is often simplified in terms of the Storage Capacity as listed in Table 1. This is calculated as the difference between the soil's Field Capacity and Wilting Point. Field Capacity correlates closely to the air entry value but is generally considered to be the volumetric moisture content of a soil at a soil suction of 33 kPa. Likewise, the Wilting Point correlates closely with the residual suction, but is generally considered to occur at a suction of about 1,500 kPa. The Field Capacity is the point where a soil starts to de-saturate, and the Wilting Point is the point where plants are no longer able to extract moisture from the soil pore space. Generally the greater the storage capacity of a soil, the greater the potential for being an efficient store-and-release cover, provided the climate is suitable.

Lastly, a soil cover must resist surface erosion and therefore silty soils, which are highly prone to erosion, are normally considered less desirable.

It can therefore be concluded that the available soils at Minto are not ideal store-and-release materials (although certainly workable) and are definitely not suitable for construction of low

infiltration (barrier) covers. Construction of such covers will require intervention such as adding bentonite.

2.4 Waste Settlement

Mine waste settlement, specifically differential settlement can affect cover integrity. The waste rock dumps and dry-stack tailings at Minto are not expected to undergo significant settlement; however, the tailings in Main Pit will undergo settlement as a result of self-weight consolidation, as well as the surcharge from cover placement.

Consolidation testing has not been done on the Main Pit tailings; however, based on general experience with similar tailings it is conceivable that tailings consolidation, and associated cover settlement could be around 30 cm on the exposed beaches, and up to 1 m on the slimes regions near the central pond area (not accounting for entrained ice). Such settlement could affect cover integrity, and therefore the cover design should either compensate for potential settlement, or construction should only be done once settlement is complete.

Complete tailings settlement will require tailings dewatering, and even then it could take years or even decades. Therefore, it is recommended that the Main Pit tailings impoundment cover design be such that up to 1 m of settlement would still result in the cover performing as designed.

2.5 Seismicity

According to the 2010 National Building Code of Canada seismic hazard calculator, the corresponding peak ground acceleration (PGA) for the Minto site is 0.057 g for a 2% probability of exceeding in 50 years (1 in 2,500). This means that this site is not particularly seismically active.

Assuming the cover is constructed from a non-liquefiable material, seismic action will affect cover integrity in two ways:

First, a cover placed on steep side slopes not subject to liquefaction, would be subject to classic failure mechanisms such as increased pore water pressures induced by ground shaking. At the Minto site this failure mechanism would only apply to covers constructed on the waste rock pile and dry-stack tailings impoundment.

Second, a cover constructed on potentially liquefiable material such as the Main Pit tailings could cause surface manifestations of liquefaction in the cover in the form of cracks and boils.

The occurrence of such surface manifestations, assuming liquefiable tailings at depth, is a function of the thickness of the non-liquefiable cap (cover plus the unsaturated tailings layer). For a cap thickness greater than 3 m, case histories (Ishihara 1985, as reported by Ritchie (1999)) suggest that there will not be a surface manifestation for ground surface accelerations up to 0.2 g. Since this is an order of magnitude greater than the design earthquake this is not considered to be a concern.

2.6 Trafficability/Constructability

The construction of any cover over saturated tailings is challenging, due to the fact that construction equipment cannot travel over the surface that needs to be covered. This means that the cover has to be constructed in one of five possible ways:

1. Construction of the tailings cover is delayed until the tailings have had enough time to naturally dewater such that equipment can safely travel on it. The problem with this approach is that this process can take very long, i.e. many years, if ever.
2. The tailings are actively dewatered through installation of wick drains or other similar dewatering devices. This can be cost prohibitive, and still must address the issue of access onto the tailings for the installation of these dewatering devices.
3. A platform is developed from the perimeter of the saturated zone by dumping cover material and dozing it over the saturated tailings. This platform becomes the working base for the construction equipment as it advances. The problem with this method is the amount of cover material that is required is significantly greater due to the requirement for a trafficable surface, and often a tailings bow wave forms immediately ahead of the working face. This results in an uneven tailings surface which may add to the need for increased cover material.
4. Cover construction can sometimes be done with specially modified low ground pressure equipment. This still requires a de-saturated surface layer, and the equipment is highly specialized and expensive.
5. Construct the cover in winter, after the frost has penetrated to a depth sufficient to support construction equipment.

Cover construction at Minto over the slimes zones of the Main Pit tailings impoundment will definitely be subject to trafficability challenges. Therefore, any cover design does need to mitigate these challenges, for example by doing winter construction to take advantage of a frozen tailings surface.

For the dry-stack tailings impoundment and the waste rock piles cover constructability should not be too much of a concern, provided that where possible the side slopes be kept flatter than 33% (3H:1V). Cover placement on steeper slopes becomes less efficient. Should geosynthetics be used, slopes may have to be flattened to as low as 20% (5H:1V).

2.7 Physical Exposure of Mine Waste

Physical exposure of mine waste is a human and terrestrial health and safety concern at mine sites (including Minto). Exposure pathways include direct physical contact, as well as indirect contact via dust and overland surface runoff. In order to mitigate this, any physical separation cover over the mine waste areas would suffice. There may however be other cover design criteria that could dominate such as infiltration control, and therefore a cover for the sole purpose of separation may not be appropriate at Minto.

2.8 Oxygen Reduction

As previously documented, geochemical characterization of the tailings and waste rock suggest that rapid oxidation is not a concern. Therefore construction of oxygen limiting/reducing covers at Minto is not warranted.

2.9 Infiltration Reduction

It has already been documented that the available cover materials are not good candidate materials to construct effective barrier covers, even if well compacted. Furthermore, in most cases, the saturated hydraulic conductivity of the mine waste is less than or equal to the available cover material, further negating the possible usefulness of constructing barrier covers with these candidate cover materials. Although data is not available it is estimated that the saturated hydraulic conductivity of the dry-stack tailings is about 1×10^{-7} m/sec (3,154 mm/year), and the Main Pit tailings (beach tailings) would be similar. The slimes would be at least an order of magnitude less (i.e., 1×10^{-8} m/sec or 315 mm/year). Data for the waste rock is not available but it is likely to be about 1×10^{-4} m/sec (3,153,600 mm/year).

Site specific water balance calculations suggest that about 30% (Clearwater 2010) of precipitation gets discharged as overland runoff at the Minto site. Very little overland runoff is however observed and, in reality, the bulk of this volume is actually shallow infiltration which emerges as stream flow downstream of the Minto site at the flow gauging station. Therefore it is not unreasonable to assume, in the absence of specific infiltration analysis, that although the available soils would allow for construction of store-and-release covers, they would likely not be able to consistently perform to a standard better than reducing infiltration to about 20% of mean annual precipitation.

As a result, in all likelihood, the only viable method to reduce infiltration further would be to make use of a synthetic cover, or alternately a bentonite-amended barrier layer.

2.10 Slope Stability

For the most part, the cover designs at Minto will be on relatively flat surfaces. The waste rock piles and dry-stack tailings impoundment have been designed with final slopes of either 2.5H:1V or 3H:1V which would support efficient cover construction, provided the slopes are not too long. Should geosynthetic covers be required, these slopes may be considered too steep.

2.11 Wind Erosion

The fines content of the candidate cover material suggest that, under prolonged dry periods, wind erosion could be a concern with respect to cover integrity. Appropriate mitigation strategies could include sacrificial increase of the cover thickness to compensate for material loss over time, or providing a physical barrier against wind erosion.

Wind erosion is a complex phenomenon and is a function of the soil properties, climate and the vegetation characteristics. Wind erosion simulation models can be used to evaluate what the potential soil loss would be; however, the lack of data precludes the use of such models at this time.

A number of physical wind erosion studies provide some useful indicative data to use as a first order estimate of wind erosion rates in this conceptual design. Forward *et al*. (2004) reported that the annual soil loss through wind erosion from agricultural land in Southern Australia ranges between 0.01 and 0.06 mm/year. Basher and Webb (1997) reported wind erosion rates in bare soil in New Zealand of 0.90 mm/year. Based on this data, a sacrificial increase in soil thickness for a 200 year cover design life would be between 3 and 180 mm. However, considering the uncertainty associated with these estimates, the difficulty in transposing the data to a site specific condition, combined with the risk of exposed mine waste, mitigation against wind erosion through sacrificial increase of the cover should be considered a potential optimization strategy at this time.

An alternative method of preventing wind erosion would be to construct a physical barrier (i.e., some form of armouring such as fine gravel) on the erosion susceptible soil cover. Since there is no readily available source of such a material at Minto, this mitigation measure is likely not viable.

Cover stabilization using vegetation is commonly acknowledged and is likely the preferred method of wind erosion stabilization for the site.

2.12 Overland Surface Runoff

The high silt content of the cover soils suggest that surface erosion will be a problem. Just as with wind erosion, overland surface water erosion rates are difficult to estimate, and can be managed by adding sacrificial cover material, or providing a suitable physical barrier. The thickness of a sacrificial cover layer cannot be estimated at this time; and further optimization in this area is recommended.

Similar to wind erosion, a definite method of ensuring that erosion protection is provided in the long-term would be to clad the erosion susceptible cover with erosion resistant soil or rip-rap. Since there is no readily available material of this nature at Minto (other than waste rock), this alternative is not recommended.

Vegetation is a proven erosion stabilization technique and is therefore the preferred mitigation strategy against overland surface runoff. To further reduce the risk of developing erosion gullies, appropriate landform engineering must be carried out to promote the development erosion resistant landforms.

It is important to remember that water shed by runoff has to be safely conveyed over the cover to its receiving environment. Therefore, the cover design must account for the volume, frequency and intensity of runoff that is anticipated and the seamless integrated designs of the cover and conveyance channel contact zones are integral to the overall success of the cover.

Naturally, more runoff implies that less water is available for infiltration. This presents potential opportunities that are of relevance in selecting a suitable cover design for the Minto site. Firstly, if a significant portion of the freshet season infiltration is shed as runoff, then the amount of infiltration that has to be accounted for in a store-and-release cover becomes less, and the likelihood of reaching a workable ratio of infiltration versus potential evaporation increases (although it may still not be enough), making this type of cover more viable.

2.13 Evapo-Concentration

Chemical constituents present in the mine waste pore water, specifically salts, can be transported up ("wicked") towards the waste surface, and continue up through the cover soil due to capillarity associated with upward fluxes caused by evaporation. This problem, also termed evapo-concentration, is common in net negative climates where the predominant flux is upward. If evapo-concentration occurs, it leads to contamination of surface water, which would constitute as non-compliance with the site closure objectives. During the wet season, there would be a constant variability of the flux within the cover, but evapo-concentration is not likely to be a major concern. However, during prolonged dry periods, the predominant flux would be upwards, especially if the underlying waste has a high moisture content. As a result evapo-concentration is not expected to be much of a concern at Minto.

This can however be mitigated by making the cover thick enough that all the meteoric action occurs within the cover with no contribution from the underlying waste, or by including a physical barrier in the cover that prevents this pore water from migrating up through the cover. The most common physical barrier is a capillary break. Since evapo-concentration is not expected to be a concern at Minto, the cover design will not take this into consideration.

2.14 Root Uptake

Allowing vegetation to establish on the final Minto closure covers is probably the most sustainable long-term solution, notwithstanding the limited growing season. A vegetative cover offers benefits in terms of reducing meteoric infiltration through increased evapotranspiration, whilst also providing erosional stability from wind and surface runoff. Disadvantages of a vegetative cover include creation of preferential flow paths through root penetration, and potential contaminant uptake from the underlying waste by the plant root systems to the surficial plant biomass, which in turn could pose a terrestrial exposure risk.

Mitigation against root uptake could include limiting plant growth outright, or constructing a chemical or physical root barrier as part of the cover. Outright prevention of vegetation can only be done through active site management, since even if coarse rock is used as a cover, dust and seed will be transported to the cover over time through wind from neighbouring undisturbed land and, ultimately, some vegetation will establish. Since active site management is not acceptable, a chemical or physical root barrier should be included as part of the cover.

Chemical root barriers, although expensive, work well in cold climates and are likely to have a significant lifespan. A more common root barrier would be a physical barrier, consisting of a layer of gravel and cobbles filled with fines such as the candidate cover soils. The fines will prevent the

layer acting as a capillary break. This root barrier should be at least 20 cm thick and should be above the barrier layer or, alternatively, below the store-and-release cover layer.

The root barrier will serve a secondary purpose of acting as a bio-intrusion layer preventing burrowing animals from penetrating the underlying waste. A root barrier needs to be constructed at Minto if low infiltration covers are required and active vegetation management is not being proposed.

2.15 Animal Activity

Minto is in a wilderness area and, therefore, after closure terrestrial animals are expected to move across the site. Burrowing activity by animals could compromise the cover, and to prevent such an occurrence, a gravelly layer can be designed into the cover. The likelihood of large scale damage due to animal burrowing action is very small and, therefore, such measures are not recommended at Minto.

2.16 Human Activity

Following closure, the site is likely to be infrequently used by seasonal hunters. Hunters typically use snowmobiles and all-terrain vehicles to get around and these machines can erode away the vegetation layer of a cover and subsequently result in damage to a geosynthetic liner. This can be mitigated by limiting access to the covered areas or, alternatively, adding in a coarse rock protection layer to the cover. The site is, however, sufficiently remote that concentrated access routes from these off-road vehicles are not expected and therefore no specific design elements will be included to protect against such damage.

3 Generalized Cover Types

3.1 Water Covers

Water covers involve permanently submerging mine waste. A water cover effectively shuts down oxygen ingress to the underlying waste, preventing oxidation. The water cover provides a constant hydraulic head that promotes seepage which may provide a pathway for mobilizing soluble oxidation products present in the waste. Therefore, water covers work best if oxidation of the waste is prevented from starting in the first place. If the waste has already undergone significant oxidation, exclusion of oxygen through a water cover to prevent further oxidation may not be warranted, or even desirable, given the inventory of oxidation products already present in the waste.

Water covers are best suited to net positive climatic water balance areas (i.e., annual precipitation exceeds evaporation), since that implies a surplus supply of water to the cover. Water covers are less favourable when permanent large engineered water retaining structures are required, such as dams that may be subject to long-term integrity concerns.

Based on the diffusion coefficient of oxygen through water, it is accepted that under ideal conditions only 30 cm of water cover would be sufficient to prevent oxidation. In reality, waste disposal facilities are typically large, and are subject to wind induced wave action, counter-current flows, seiching, lake/pond turnover (and lake/pond ice formation in cold climates) etc. These physical actions result in water turbulence, which in turn, may result in re-suspension of particles. Re-suspended particles may undergo oxidation rendering the water cover ineffective. Rules-of-thumb state that a 1 m thick water cover would mitigate against these concerns; however, a review of current practice shows that typical water cover depths are between 2 and 5 m thick.

As described in Section 2.2, the climate at Minto is conducive to the use of water covers. The mine waste at Minto is however not highly oxidizing and therefore maintaining a permanent water cover is not necessary for long-term geochemical stability. More importantly however is the fact that, with the exception of tailings deposited in Main Pit (and possible future pits), the mine waste rock and dry-stack tailings facility cannot readily be flooded unless large containment structures are constructed, and/or waste is relocated. For the tailings in Main Pit, a water cover could likely easily be engineered; but a perpetual water retaining dam would be required. Water covers were, therefore, not given further consideration at this time.

3.2 Saturated Soil Cover

Saturated soil covers are an alternative method of ensuring isolation of wastes from exposure to oxygen, but by eliminating the need for large stretches of open water, the issues associated with re-suspension are eliminated. The concept entails using a coarse (ideally gravelly) material with significant void space as the primary cover material and placing it in a perpetually saturated state. This saturated layer can be at surface or at depth, i.e., providing opportunity for an upper cover layer that may, for example, sustain vegetation. This type of cover works best if the climatic water balance is positive, and just as with the water cover, may be problematic if significant oxidation products are already present.

At Minto this type of cover may have merit for in-pit tailings although, given the fact that neutral metal leaching is a concern, the constant hydraulic head and subsequent risk of seepage may be problematic. Therefore this cover type was not explored further at this time.

3.3 Barrier Covers

3.3.1 Concept

Barrier covers (also known as water shedding or infiltration controlling covers) work on the principle of physically limiting and/or preventing meteoric infiltration and/or oxygen ingress. As described above, if the waste has been significantly oxidized, an oxygen barrier may be of limited benefit; however, a cover controlling infiltration may be of benefit.

Physical barriers can come in many different forms, including impermeable synthetic barriers such as high density polyethylene (HDPE) liners, low density polyethylene (LDPE) liners, bituminous liners, geosynthetic clay liners (GCL), or natural low permeability soil. A barrier cover can also be constructed by amending the surface layer of mine waste to achieve a low permeability layer. A brief summary of these different barrier cover types is provided below.

3.3.2 Natural Low Permeability Soil Cover

The effectiveness of a natural low permeability soil cover is determined by the following criteria:

- The cover should have a sufficiently low saturated hydraulic conductivity in order to restrict meteoric infiltration rates to the underlying waste;
- The cover should have a significantly lower saturated hydraulic conductivity than the underlying waste; and
- The cover should maintain a high degree of water saturation in order to limit the entry of atmospheric oxygen to the waste (a rule of thumb is at least 85% saturation).

A natural low permeability soil cover that is intended to act as both a water and oxygen reducing cover should adhere to all criteria above, whilst if the cover is only required to act as an infiltration barrier it needs to only adhere to the first and second criteria. If the intent of the cover is to act as only an oxygen barrier, it needs only to adhere to the third criterion.

Natural low permeability soil covers work best in net positive water balance climates which are not subject to freeze-thaw cycles. Under such conditions, there is sufficient moisture to keep the level of cover saturation high, and cover degradation due to frequent wetting and drying cycles (i.e., shrinkage and cracking) is limited. The annualized climatic water balance at Minto is largely net positive; however, three months of the year it is net negative, and more importantly, the site is subject to annual ground freezing. Therefore, if the barrier layer can be constructed below the annual depth of freezing, a barrier cover would likely work well at the site.

Generally, natural low permeability soil covers consist of clays and silts, with plasticity ranging from moderate to high. These materials are generally placed in a compacted state. Theoretically, a low permeability cover does not have a minimum thickness, i.e., the performance of a 1 m thick layer with a specific saturated hydraulic conductivity value would have exactly the same net effect as a geosynthetic clay liner (GCL) of the same saturated hydraulic conductivity given the same hydraulic gradient (i.e., unity). In practice, however, low permeability layer thickness is governed by constructability and allowance for desiccation cracking. Therefore, if the low permeability cover: (a) has limited desiccation cracking risk due to a well-graded texture and low plasticity, (b) is not subject to extreme wetting and drying cycles; (c) is below the depth of ground frost penetration; and, (d) is protected against surface runoff erosion, then the cover could be relatively thin, and constructability issues would be the primary concern (i.e., it is difficult to place a consistent thin compacted layer of low permeability soil over an undulating waste pile surface, and it is almost impossible to do so on soft saturated ground).

It is generally considered poor practice to allow vegetation to establish directly on a compacted low permeability layer because the design hydraulic conductivity and performance will be compromised. If a single-layer cover is adopted, using only the compacted low permeability cover material, then the thickness must be appropriately increased to allow at least a 60 cm thick layer of compacted low permeability soil that will not be penetrated by roots. As an added safeguard, consideration can be given to placing a root barrier. At Minto, use of a single layer at surface is not recommended due to ground freezing.

In order to get a general sense of what type of cover material would be required to perform as an effective natural low permeability soil cover at the Minto site, some general calculations can be carried out. Infiltration due to precipitation will be reduced if the hydraulic conductivity of the soil cover profile is less than the intensity of precipitation. For example, if the annual precipitation at Minto is about 335 mm, or 1.06×10^{-8} m/sec, water infiltration will be limited should the hydraulic conductivity of the cover be less than this value. Therefore, if it is intended to reduce infiltration through the cover to less than 20% of mean annual precipitation (i.e., less than 67 mm), the cover material should have a saturated hydraulic conductivity of at least 2.12×10^{-9} m/sec. From a water infiltration perspective, this a conservative assessment, since in reality the cover would not be completely saturated all the time, and based on unsaturated soil property theory, the hydraulic conductivity of an unsaturated soil profile is significantly less than in its saturated state. However, if the profile is significantly de-saturated to benefit from a low unsaturated hydraulic conductivity, the soil air entry value will be exceeded, and significant oxygen ingress will occur, i.e., the cover cannot be an oxygen barrier.

Current soil characterization data suggest that the saturated hydraulic conductivities of the available candidate cover materials range from 1.99×10^{-7} to 1.12×10^{-7} m/sec (6,275 to 3,532 mm/year), which is clearly not suitable for use as a barrier cover. Experience has shown that in practice, it is not realistic to expect natural soil barrier covers to have saturated hydraulic conductivities of less than 10^{-7} m/sec (about 3,154 mm/year), even though laboratory tests would often suggest otherwise. This is especially true if the barrier layer is located within the soil active zone (i.e., it is within the zone of cyclic wetting and drying, or freezing and thawing). Furthermore, Wilson *et al*. (2003) demonstrate that world wide experience with natural soil barrier covers,

especially in seasonally wet/dry climates almost always yield greater than expected or predicted infiltration numbers. Given the site climatic conditions, SRK would not recommend the use of natural soil barrier covers at Minto using the locally available soils.

3.3.3 Synthetic Cover (HDPE, LDPE, Bituminous Liner)

Low permeability soils, if available, are almost always more economical than synthetic products; however, when natural soils are not readily available, or if the environmental risk is high, synthetic products such as capping materials can be appropriate. HDPE, LDPE, and bituminous liners are essentially impermeable; however, due to manufacturing and installation errors, even under the best quality control procedures, small pin holes can develop in the liner, such that generally for design purposes these liners are assumed to have saturated hydraulic conductivity values in the order of 10^{-15} m/sec (0.00003 mm/year). The difference between HDPE, LDPE, and bituminous products is generally the chemical resistance and workability of the material, with HDPE and bituminous liners being more robust products.

Synthetic liners must be installed according to manufacturer installation procedures that call for seaming of joints between liner sections, as well as placing the liner on an appropriately prepared surface. An appropriately prepared surface implies a clean, stable, smooth surface with limited undulations and few surface protrusions. If the waste surface is sufficiently coarse, the liner may be damaged, and an appropriate bedding material consisting of either a layer of clean sand or a non-woven geotextile are required.

Finally, the liner would have to be covered with a suitable cover depending on the final adopted land use. Again, if the cover soil contains a significant coarse fraction, then a protective sand or geotextile layer must first be placed over the liner.

An HDPE, LDPE, or bituminous liner would effectively eliminate oxygen and water ingress and provide a reclamation alternative with the least risk of impacting the environment in the long-term. Synthetic liners do, however, have a finite lifetime (manufacturer's guarantee is normally limited to about 20 to 40 years), and there is a possibility that the closure plan would have to allow for complete liner replacement every 100 to 200 years. Finally, synthetic liners cannot be constructed on steep slopes, with the general rule of thumb being slopes flatter than 3H:1V for short slopes, and as flat as 5H:1V for longer slopes. If low infiltration covers are required at Minto, these types of covers should be considered.

3.3.4 Synthetic Cover (Geosynthetic Clay Liner)

A GCL consists of a thin layer (generally less than 1 cm thick) of bentonite clay sandwiched between two non-woven geotextile layers. Under ideal conditions, a GCL is generally assumed to have a hydraulic conductivity of approximately 10^{-11} m/sec (0.3 mm/year); however, in practice a value of 10^{-9} m/sec (31.5 mm/year) is often adopted for design calculations. GCLs are considered to be less susceptible to installation damage than HDPE, LDPE, or bituminous liners; however, they must be installed according to manufacturer's installation procedures. Since GCLs are generally used with liquid containment in mind, and the use of a GCL as a cover (infiltration barrier) is somewhat different, the sub-grade specification can be relaxed. Depending on the

surface preparation, it is conceivable that the GCL may be placed directly onto the mine waste and similarly the protective cover could be placed directly onto the GCL. The built-in geotextile, if appropriately specified, may provide sufficient protection.

A cover which includes a GCL may prove to be an extremely effective infiltration and oxygen barrier, provided the bentonite can remain hydrated and confined at all times. At the Minto site, this hydration should not be a problem during the wet season, but during the dry season, the bentonite may de-hydrate if the protective layer does not have sufficient moisture retaining capability. In addition, if the GCL is subject to freeze-thaw cycles, the physical integrity of the liner may be impacted. A GCL can be penetrated by roots, and therefore, as with a natural low permeability soil cover, it is desirable to ensure the vegetative substrate is thick enough to ensure roots do not reach the GCL, or alternatively, introduce a root barrier into the cover design.

It is also important to carefully consider the calcium content of meteoric water that will permeate through the GCL. As water passes through the GCL, calcium will cause ion exchange with sodium in the bentonite, which leads to permanent aggregation of the GCL. This results in an increased hydraulic conductivity, and subsequently increased infiltration rates (Lin and Benson 2000).

GCL covers are worth considering at Minto if low infiltration covers are required.

3.3.5 Bentonite Amended Mine Waste/Cover Soils

It is possible to reduce the surface hydraulic conductivity of the mine waste or a candidate cover material by mixing in bentonite clay and compacting the mixed layer, thereby creating an in-situ low permeability layer. This type of amendment is often done as a more cost effective solution than a synthetic cover to reduce surface permeability due to the fact that limited importing of soils and/or synthetic products is required. The hydraulic conductivity is reduced by mixing in bentonite clay which has a high affinity for moisture and hydrates to more than twice its volume, thereby taking up void space in the host material, which would otherwise have a high hydraulic conductivity. Naturally the same restrictions apply to bentonite amended soils with respect to possible aggregation of the bentonite as a result of ion exchange as has been previously discussed for GCLs.

There are two primary methods to mix the bentonite and host soil (or mine waste): (a) off-site mixing using a pug mill, and (b) in-situ mixing. Off-site mixing requires stockpiles of host material and bentonite, which is fed into a pug mill, and the mixed product is trucked to the deposition location for application and compaction. This method allows for excellent control over the mix, including moisture control which is important to ensure proper hydration of the bentonite (the mix is normally placed at above optimum moisture content). In-situ mixing requires the upper layer of the location to be ripped before spreading the bentonite at the desired application rate.
The bentonite is then mixed into the host soil using agricultural equipment. This method allows for less control over application rate and moisture, and therefore requires more bentonite product. However, if the host material is not pre-stockpiled then this method is usually more cost-effective.

The thickness of the bentonite amended layer will be site specific; however, a realistic thickness for use as a low permeability cover would be between 15 cm and 30 cm thick, with an application rate of 4 to 6% bentonite by weight (of the host soil). The bentonite-amended layer will perform as an effective low permeability layer, reducing infiltration and oxygen ingress as long as the bentonite remains hydrated. Should the layer be allowed to dry out, the volume change will result in cracks, which will increase the hydraulic conductivity. Bentonite does however have self-healing characteristics which will help cracks to reseal if the bentonite becomes re-hydrated. Normally, to minimize chances of drying of the bentonite amended layer, it would be overlain with 50 cm of compacted, but non-modified material that would remain at high saturation due to the low permeability of the bentonite amended layer. Should vegetation be a requirement, a suitably designed vegetation substrate layer would have to be placed on the amended and compacted layers, to limit root penetration of the amended layer. Bentonite amended covers would be a worthwhile alternative to consider at Minto should low infiltration covers be required.

3.4 Store-and-Release Cover

An alternative method of reducing infiltration is to make use of the inherent soil moisture storage capacity of the soil. Meteoric infiltration is not prevented as with a barrier cover, but rather promoted, such that the moisture is stored in the cover soil. This moisture is returned to the atmosphere via evapotranspiration before is has the opportunity to drain through the cover and enter the underlying waste. These covers are referred to as “store-and-release” covers (also called “water retention” or “alternative” covers), and work most effectively in dominantly net negative climates, i.e., the potential evaporation greatly exceeds the mean annual precipitation (typically at least two or three times greater).

As discussed in Section 2.2, the site climate is not particularly well suited towards use of store-and-release covers; however, during the summer months, they certainly can be effective provided that the wet (frozen) season precipitation is shed as runoff, such that the amount of infiltration for which storage would have to be created, would be manageable. It may also be beneficial to provide a barrier layer beneath the store-and-release layer to prevent breakthrough during the prolonged wet season.

Ideal store-and-release cover soils are well-graded soils that have sufficient fines to allow capillarity to be high whilst not compromising too much on the material porosity. Store-and-release covers are less susceptible to site preparation and quality control since compaction of the soil cover is generally not required (or in fact not recommended) and the cover is of sufficient thickness to mask minor undulations on the waste surface. Settlement of the waste will also not affect the cover as much. Direct re-vegetation of the store-and-release cover is generally undertaken, unless the soil has a nutrient deficiency. The successful establishment of vegetation on a store-and-release cover is crucial towards its success, since the enhanced evapotranspiration will ensure long-term cover performance.

Notwithstanding the reservations about the applicability of using store-and-release covers at Minto, we can use the estimated hydraulic properties of the available cover soils at Minto, and use empirical methods (Chen 1999) to estimate the likely range of store-and-release cover thicknesses that would be required at Minto. Assuming 20% mean annual precipitation as the

infiltration target, the store-and-release layer thickness would have to be between 384 and 740 mm. This increases to between 676 and 1,225 mm if the infiltration target was lowered to 2%. It is important to note that due to the relatively low air entry value of these soils, their capillarity is limited and, therefore, increasing the thickness of the store-and-release layer provides very little added benefit. Although the available storage space is increased, the moisture cannot be moved upward more than about 1 m at most, but more importantly, there is not enough radiating energy available to extract the excess moisture.

At Minto, considering the available soils and climate, an infiltration target of about 20% on average for a store-and-release type cover is probably as good as would be consistently attainable. Under specific conditions this cover is likely to yield much better performance; however, significant breakthrough events will frequently occur.

3.5 Reactive Cover

Reactive covers normally refer to covers that include chemically active components. Broadly speaking, two types of such reactive covers are referred to: (1) covers where the reaction leads to consumption of oxygen, and (2) covers where the reaction leads to formation of low permeability layers and/or barriers to oxygen diffusion.

Oxygen consuming covers can be either organic or inorganic. Organic covers, for example wood waste over an acid generating tailings impoundment have been shown to have good success, however the main complication with these kinds of covers is that they are not sustainable. The organic cover loses its ability to consume oxygen over time, and as a result become ineffective. Organic oxygen consuming covers can be in the form of materials that have high neutralizing potential, and as a result, balance out the overall seepage waste load as a result of the buffering capacity.

The concept of deliberately placing reactive materials to create low-permeability covers rests on the principal that two layers are placed in contact, each one being a reagent. Where these layers contact each other, the reactants will meet and combine to form a precipitate that fills the pores.

Although in theory reactive covers show potential, to date they have not been used in full scale mine closure applications. Some long-term and medium scale pilot scale covers have been constructed; however, the principle of using reactive covers as final closure covers has not been adopted by the industry to date.

There does not appear to be any potential cover materials that would make for good reactive cover construction at the Minto site, and therefore this type of cover was not given further consideration in this evaluation.

3.6 Capillary Break Cover

A capillary break cover can be used to limit infiltration and to prevent contaminants from a waste deposit from being wicked up into the cover soil. It works by placing a fine grained soil over a coarse soil (typically a gravely soil containing little or no fines). The upper fine soil can come close to saturation, before it allows flow to pass through the underlying gravel. Generally, if a capillary break is used to limit infiltration, it can be used to reduce the thickness of an overlying store-and-release cover layer.

A capillary break may also be useful to maintain a high degree of saturation in a cover such that it would act as an oxygen barrier. Therefore, a capillary barrier may be constructed under a low permeability cover layer to ensure that the degree of saturation remains high, and limit oxygen ingress.

Capillary barriers also work in the reverse, i.e., if there is an upward flux through the cover due to evaporative action through which contaminants in the waste body may be wicked to the surface, the capillary break will prevent this from happening. Such wicking could result in surface water contamination in runoff from the cover, and it could also lead to metal uptake of vegetation established on the cover. Wicking is generally more of a concern in a net negative climate, where the upward flux often dominates. This may only be a concern during the three dry months at Minto.

Assuming the capillary barrier has been correctly designed with the appropriate hydraulic properties (it must have a residual suction which is lower than the air entry value of the overlying fine soil), a capillary break layer has no minimum thickness. In reality, the rule of thumb is to design these layers at least 30 cm thick. Depending on the difference between the gradations of the capillary break and the overlying fine soil, a filter media or geotextile may be required at the upper and lower interfaces.

In many cases, the waste material to be covered is sufficiently coarse that, if a low permeability cover is applied, the gradation difference is sufficiently large enough that the waste itself would act as a capillary break. Capillary break covers are quite complex to design and construct, but are best used to either keep an upper or lower layer at a higher degree of saturation, or to specifically prevent upwards wicking. Neither of these conditions appears to be specifically required at Minto and therefore this cover variant was not considered further at this time.

3.7 Frozen/Thermal Cover

In cold regions, mine waste may freeze, or measures can be implemented to promote freezing. This effectively eliminates any risk of contamination from these wastes. Covers can therefore be designed to preserve the frozen waste mass. This concept, generally, only works in areas where cold continuous permafrost is already present and the covers are normally very thick (i.e., in excess of 2 m).

Minto is located in the discontinuous permafrost region of Canada and the permafrost present is relatively warm (i.e., about -1^{o}C near surface). Considering climate change predictions which are forecasting warming trends, design of a frozen cover at Minto is not considered a viable option.

4 Conceptual Cover Designs

4.1 Site Wide Closure Objectives

In accordance with the D&RP (Minto 2011), the site wide closure objectives are as follows:

- Have a closure planning process that seeks input from the Selkirk First Nations, understands the input received, and incorporates the input into closure planning decisions;
- Protect the health of people pursuing traditional activities including hunting, fishing, trapping, camping, and collection of plants for food, medicinal or cultural purposes;
- Protect people from safety risks when they are pursuing traditional activities including hunting, fishing, trapping, camping and collection of plants for food, medicinal or cultural purposes;
- Protect the environment (including land, air, water, plants, animals fish and other environmental components and their interrelationships) from long-term effects caused by the mine activities and facilities;
- Return the mine site and affected areas to a state similar to surrounding lands so that people can pursue traditional activities the same as they did before mining, including hunting, fishing, trapping, camping, and the collection of plants for food, medicinal or cultural purposes;
- Protect the environment from long-term effects caused by post-closure access to the mine area;
- Protect the environment from effects of earthquakes, floods, climate change, and other natural events on related mine structures;
- Have effective management and control structures in place during operation, closure and post-closure to provide;
  - Adequate financial resources to carry out all closure activities including plan implementation and long-term activities;
  - Adequate flexibility during closure and post-closure to allow adaptation of activities in order to address unexpected performance and events; and
  - Consideration of the Company's long-term desire to "walk away" from the site under conditions acceptable to Selkirk First Nations and with adequate resources provided to address long-term requirements.
- Minimize long-term activities by ensuring long-term chemical and physical stability of mining components and disturbed areas;

- Confirm the effectiveness of closure measures by monitoring the site after closure;
- Undertake mine planning incorporating progressive reclamation;
- Provide short and long-term slope stabilization and erosion control on linear and non-linear disturbances;
- Ensure the long-term chemical stability of residual mining components and minimize their effects on water quality draining the property;
- Ensure the long-term physical stability of key structures such as the waste dumps and the diversion and drainage ditches; and
- Work towards a passive closure scenario for most or all mine components.

4.2 Possible Cover Functions

Covers are one of many possible mitigation strategies that can be used to achieve site wide closure objectives. Other two common strategies include waste relocation and water treatment. If conditions are such that covers are a suitable mitigation strategy, then the specific function that the cover needs to perform in order to meet the closure objective must be explicitly stated. The following cover functions are recommended for Minto based on the current understanding of site conditions:

- Reduce infiltration to a specific target that would ensure site wide water quality objectives can be met;
- Isolate the mine waste to avoid direct contact by human or other terrestrial animals; and
- Provide a stable landform that is aesthetically pleasing.

4.3 Cover Design Criteria

The cover functions in Section 4.2 above could be considered ambiguous if they are considered in isolation and therefore these functions needs to be explicitly defined in terms of measurable design criteria. A set of example design criteria is provided in Table 4.1.

Table 2. Proposed Minto mine cover design criteria

Component	Criteria
Design Life	100 years (while it is recognized that the cover will remain in place for a longer period, design performance will be measured against this criteria)
Oxygen Reduction	Not required
Infiltration Reduction	Isolating Cover: no more than the uncovered waste; Low Infiltration Cover: no more than 10% of mean annual precipitation with no significant breakthrough events, and; Very Low Infiltration Cover: no more than 2% of mean annual precipitation with no breakthrough events
Waste Settlement	Waste Rock Piles and Dry-Stack tailings: no criteria; Main Pit Tailings: up to 30 cm on beach areas and up to 100 cm on slimes areas (cover is not expected to perform without maintenance and repair when subject to these settlements)
Seismicity	1 in 2,500 year recurrence interval (10% probability of exceeding in 50 years)
Trafficability	Concern for the Main Pit tailings area only. Mixing zone acceptable; but minimum cover design thickness to be measured from above the mixing zone
Physical Exposure	As far as practicable keep mine waste covered. In no areas may more than 1 m^2 of mine waste be exposed as a contiguous area for the life of the structure
Slope Stability	Overall factor of safety of at least 1.3 under static conditions and 1.1 under seismic loading
Wind Erosion	No visible dust up to and including wind speeds with a recurrence interval of 1:10 years, for any given duration. In no areas may overall cover thickness be less than 75% of original design thickness for the life of the structure
Overland Surface Runoff	Capable of withstanding 1:100 year, 24 hour duration storm during peak freshet with no damage. In no areas may overall cover thickness be less than 75% of original design thickness for the life of the structure
Evapo-Concentration	Not required
Root-Uptake	Not required
Vegetation	Self-sustaining vegetation cover endemic to the region within 20 years after cover construction (foreign species and fertilizers during early years acceptable if it supports the final objective)
Land Use	General wilderness area. Large terrestrial animals, birds and aquatic life will be present. Humans will travel through the area infrequently (mostly hunters and trappers). Specific measures to preclude damage to the covers due to human and/or animal use not required
Landform	Promote use of landforms consistent with the current landscape. Provide for variability on cover thicknesses and landscaping to promote establishment of microclimates and variability

4.4 Conceptual Designs

4.4.1 Isolating Cover

Figure 15 illustrates a conceptual isolating cover for the Minto site. The bottom trafficking layer of run-of-mine waste rock is only required for the Main Pit tailings. For beach areas, the trafficking layer thickness of about 0.5 m is probably sufficient, but in the slimes areas, this layer may have to be up to 1 m thick. It is recommended that these trafficking layers be constructed during the winter season when the tailings surface is frozen as the added strength will facilitate in minimizing the mixing zone which would reduce the total volume of material required. Note that tailings may migrate through the trafficking layer as traffic passes over the cover and therefore placement of this layer alone does not constitute an isolating layer.

The isolating layer will be a minimum 0.5 m thick compacted layer of local overburden soil. No specific measures need to be taken to specifically preclude compacting through trafficking across the covered surface; however, to the extent practical, to facilitate vegetation establishment, care must be taken to prevent excessive vehicle compaction. The soil cover will then be re-vegetated.

Detailed numerical analysis has not been carried out; however it is SRK's opinion based on the material characterization and site climate that this cover would consistently result in infiltration values of between 10 and 20% of mean annual precipitation. Significant breakthrough events can, however, occur based on large individual rainfall events. The cover will be susceptible to surface erosion and, therefore, it would be of paramount importance to establish stabilizing vegetation soon after cover placement. Where surface runoff concentrates, dedicated surface water conveyance channels will have to be constructed.

The performance of this cover will not be affected by freeze-thaw and wetting/drying cycles and is not significantly affected by waste settlement. Some maintenance in the form of regrading will be required if significant differential settlement is experienced to ensure that water is not trapped in ponds on the covered surface.

4.4.2 Low Infiltration Cover

A conceptual low infiltration cover for the Minto site is presented in Figure 16. The locally available soils are not suited to constructing low infiltration covers and therefore soil amendment will be required. Although no specific testing has been carried out, it is expected that a 4% bentonite mix with the locally available soils will yield an appropriate infiltration barrier to ensure infiltration remains consistently below 10% of mean annual precipitation. The amended layer should be about 0.3 m thick, be compacted and covered with a 0.5 m loosely compacted isolating layer as described in Section 4.4.1.

The trafficking layer requirements are also consistent with those described in Section 4.4.1.

The infiltration performance of this cover will be affected by freeze/thaw and wetting/drying cycles; however the self-healing properties of the bentonite should compensate against total failure. It is, however, conceivable that small breakthrough events could occur unless the upper isolating cover is increased in thickness to about 3 m to ensure the amended barrier layer is below the annual frost penetration depth.

The cover performance is sensitive to waste settlement, as large displacements of the bentonite amended layer may result in significant discontinuities in the layer which would increase infiltration. Frequent inspections of the cover will have to be undertaken over the Main Pit tailings and where large depressions are observed, the bentonite amended layer must be exposed and possibly in-filled to compensate for potential loss of performance.

On the dry-stack and possibly the Main Pit beach areas, consideration can be given to amending the upper tailings surface with bentonite as opposed to the dedicated 0.3 m thick imported overburden layer.

Appropriate trade-off studies will have to be carried out to determine the most cost effective method of mixing the amended layer.

4.4.3 Very Low Infiltration Cover

Suitable natural soils to construct a very low infiltration cover at Minto are not available and, therefore, a geosynthetic liner is recommended if this level of cover performance is required. An appropriate trade-off study will have to be carried out to determine the most economical geosynthetic product; however, for the purpose of this study, a 60 mil textured HDPE liner has been assumed. The overall conceptual cover design is illustrated in Figure 17. As described before, the trafficking layer only applies to the Main Pit tailings impoundment.

A levelling and protective layer of compacted tailings (borrowed from Main Pit beach tailings or the dry-stack tailings impoundment) at least 10 cm thick must be placed between the mine waste and the liner. Note this is not required on the dry-stack, but will be required on the Main Pit tailings since the trafficking layer will be the surface on which the liner is placed.

A similar protection layer must be placed over the liner prior to constructing the overliner soil cover. Given the relatively benign geochemical composition of the tailings, it is recommended that this layer be tailings. Should focussed characterization prove this is unacceptable, suitable clean sand would have to be imported, or alternaticely a robust (e.g., 16 oz) non-woven geotextile can be substituted.

The overliner soil cover should consist of a 0.3 m thick compacted layer (at least 98% Standard Proctor) of local overburden soils, overlain by a 0.5 m thick, loosely compacted layer similar to that recommended for the isolating cover. The lower compacted layer serves to better anchor the liner, and to provide a root barrier to resist deep penetration which may damage the liner.

Although detailed numerical analysis has not been carried out, the liner, if installed with due care, will ensure infiltration to less than 2% of mean annual precipitation with no chance of

breakthrough events due to extreme events. The upper isolating cover will be susceptible to surface erosion and would require the same treatment as described in Section 4.4.1.

The infiltration performance of this cover will not be affected by freeze/thaw and wetting/drying cycles; however the compacted layer will become less dense overtime, thereby reducing its functionality as described. A complete loss of functionality over the design life of the cover is, however, not expected.

The liner performance is very sensitive to waste settlement, and may tear or rupture under severe conditions. Frequent inspections of the cover will have to be undertaken over the Main Pit tailings impoundment in the first 5 to 20 years following cover construction. If trampolining (i.e., unsupported liner) is observed, the liner would have to be exposed, and remedial works initiated.

5 Conclusions

The hydrologic and climatic regime at the Minto site is best suited to the use of water covers or infiltration reducing covers (i.e., barrier covers). Store-and-release and thermal covers are not likely to be successful. The available overburden soils at Minto are not well suited for construction of low or very low infiltration covers.

These soils can however be used as effective isolation and vegetation supporting covers, and are expected to result in overall infiltration of between 10 and 20% of mean annual precipitation, typically. Periodic higher breakthrough events will, however, occur. Soil amendments or use of synthetic products would be required to construct low to very low infiltration covers at the site. Conceptual designs of each of these different cover variants have been proposed.

This report ***"Scoping Level Cover Assessment for Minto Closure Covers"*** has been prepared by:

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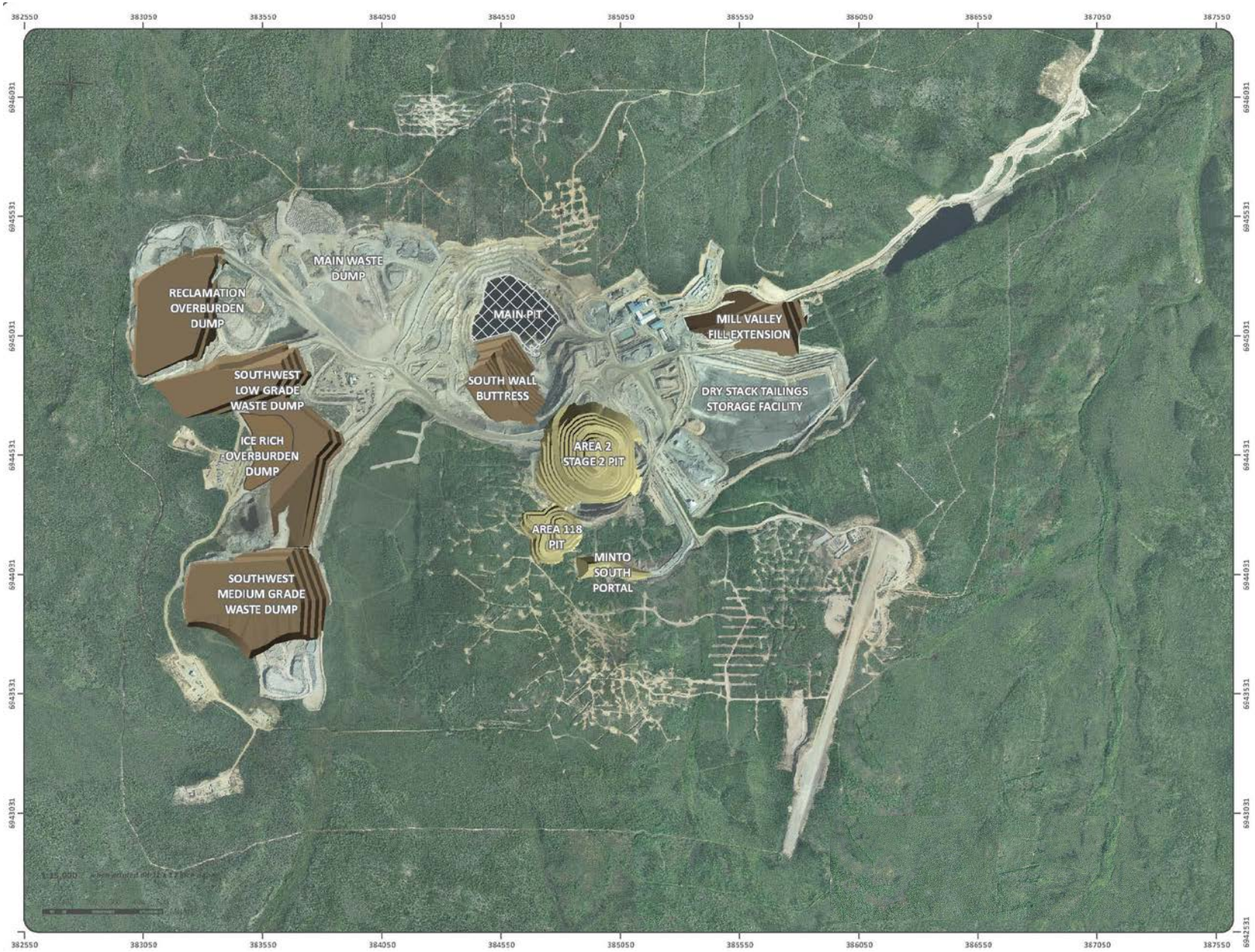
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Figures



Note: Drawing as supplied by Minto Explorations Ltd. (subsidiary of Capstone Mining Corp.)

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MINTO MINE

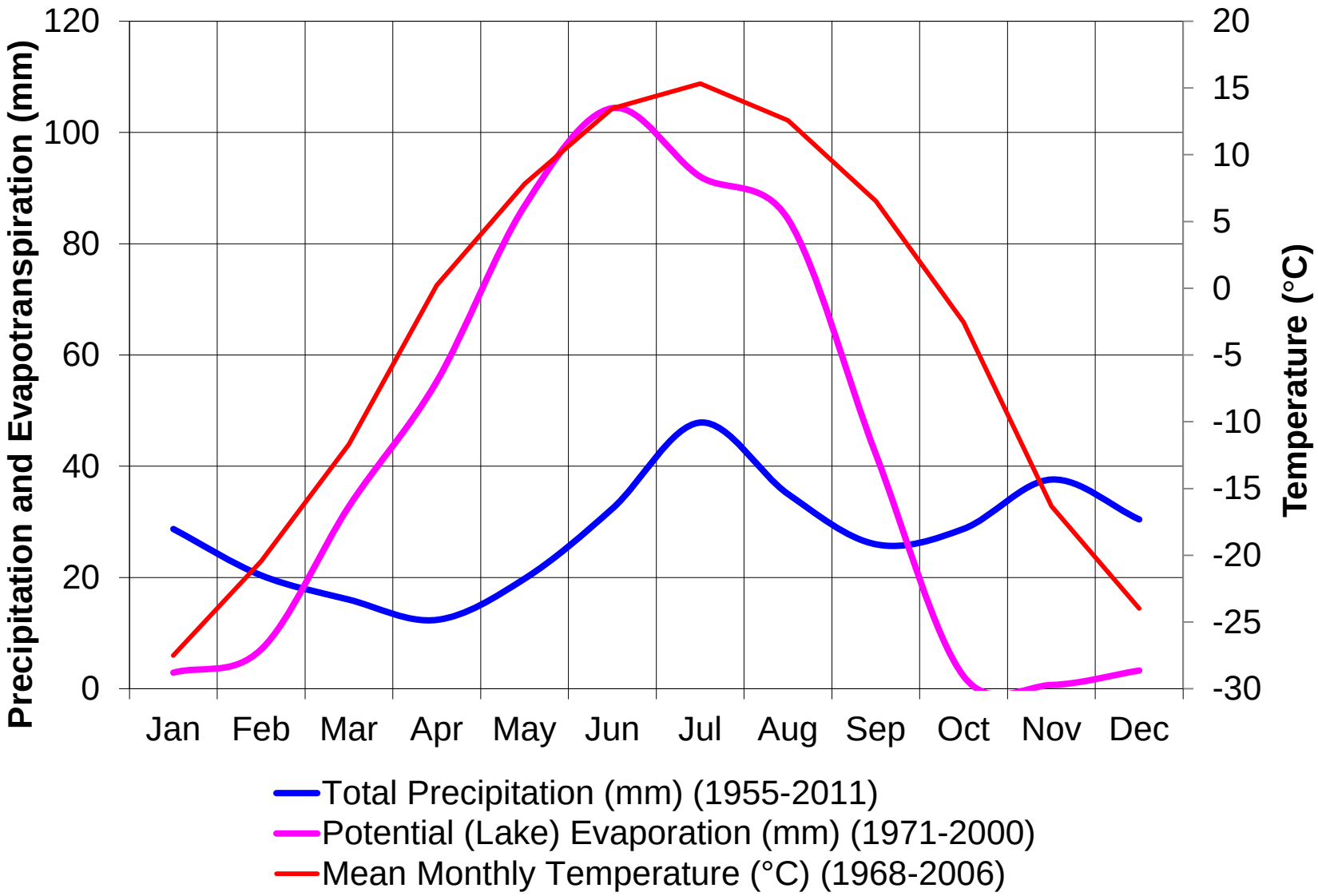
Scoping Level Cover Assessment

General Site Layout

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Minto Mine

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Scoping Level Cover Assessment

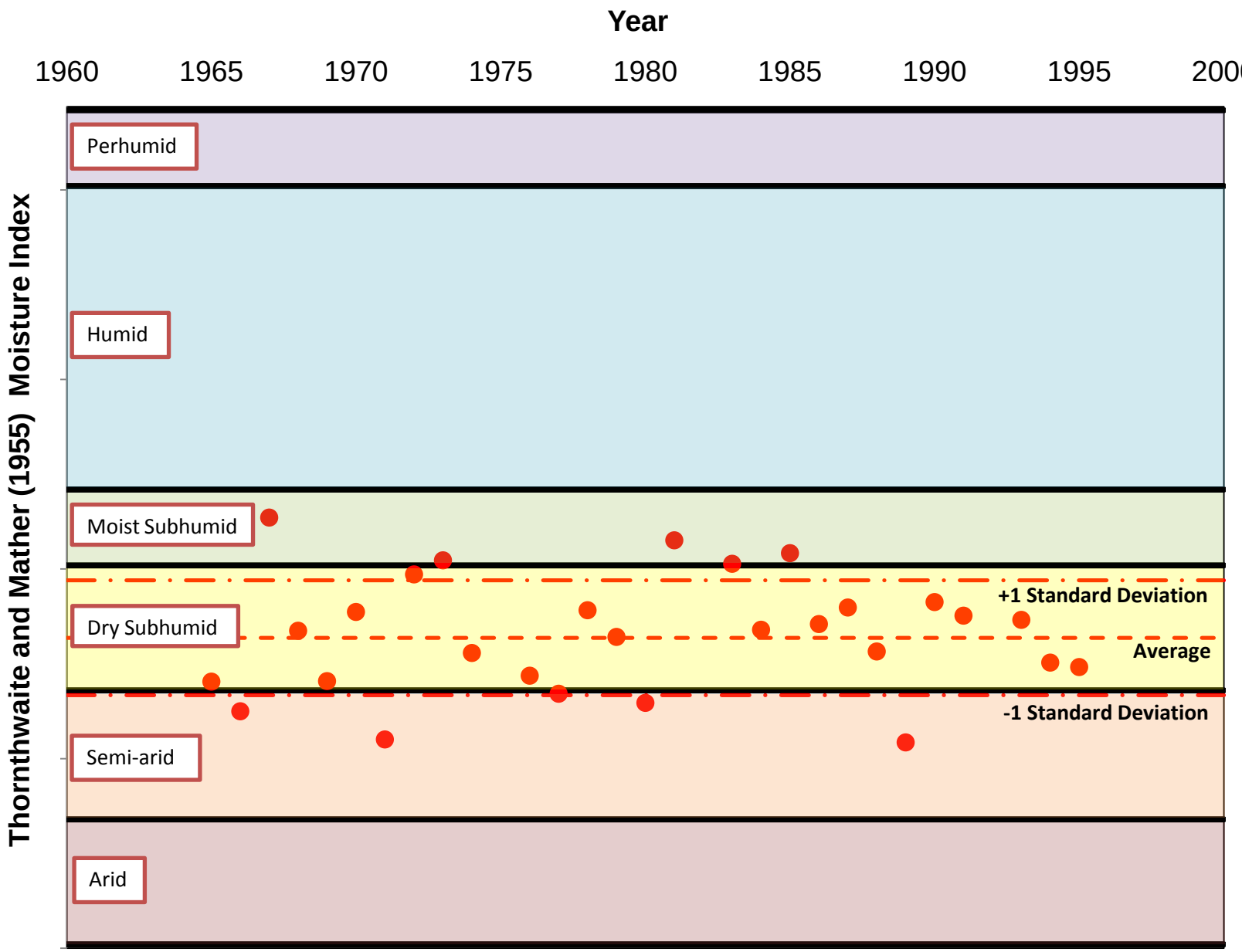
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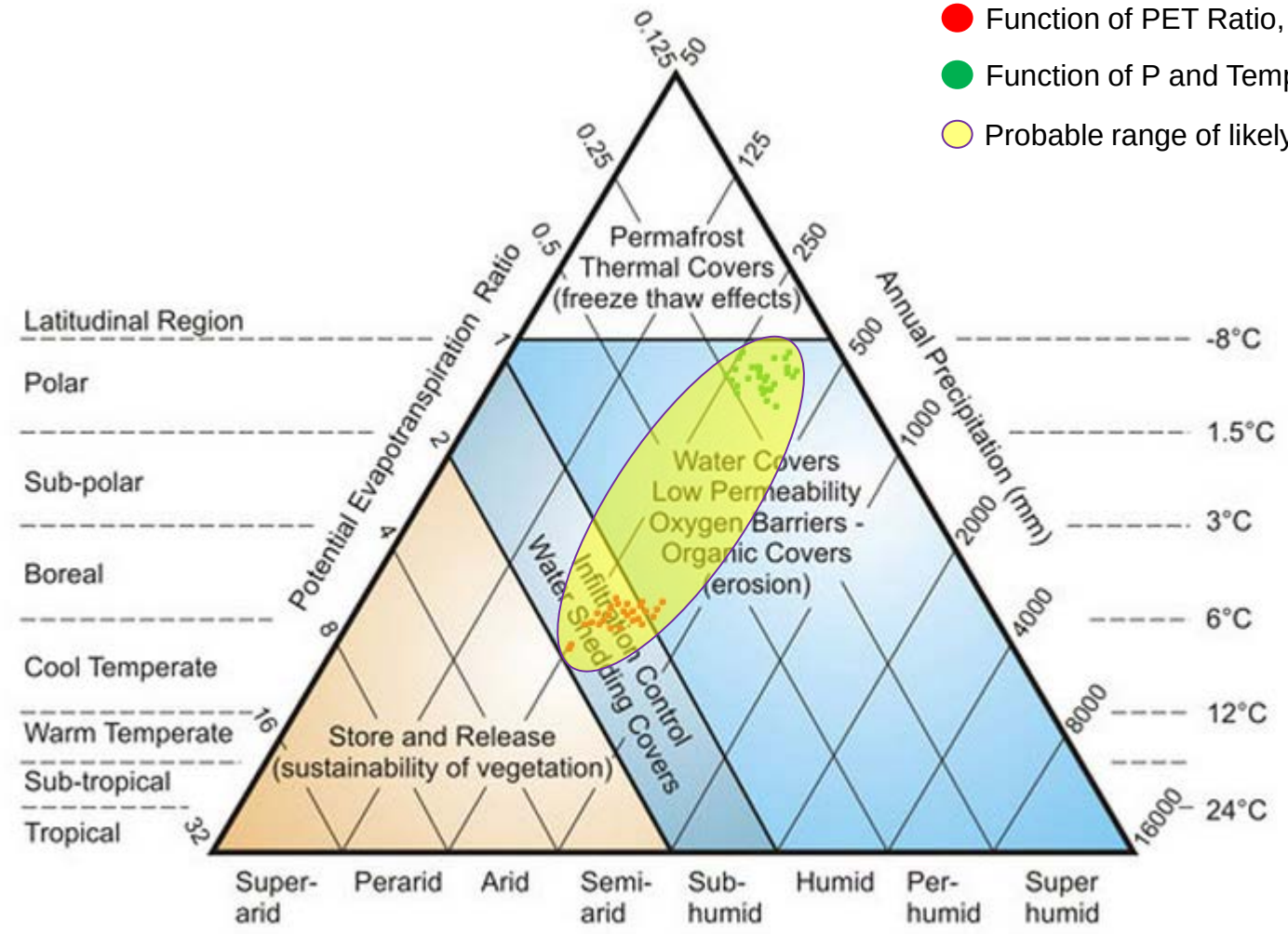
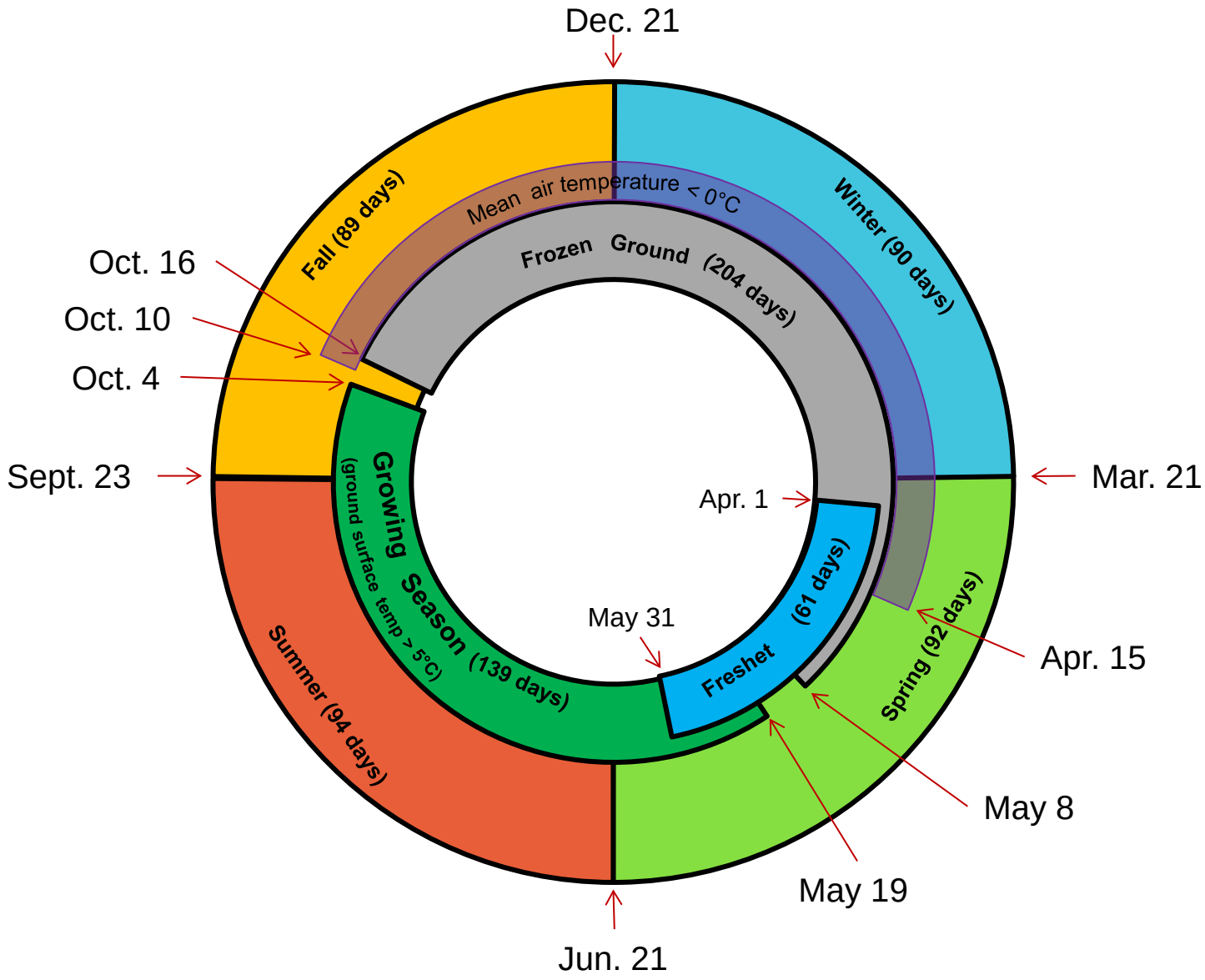


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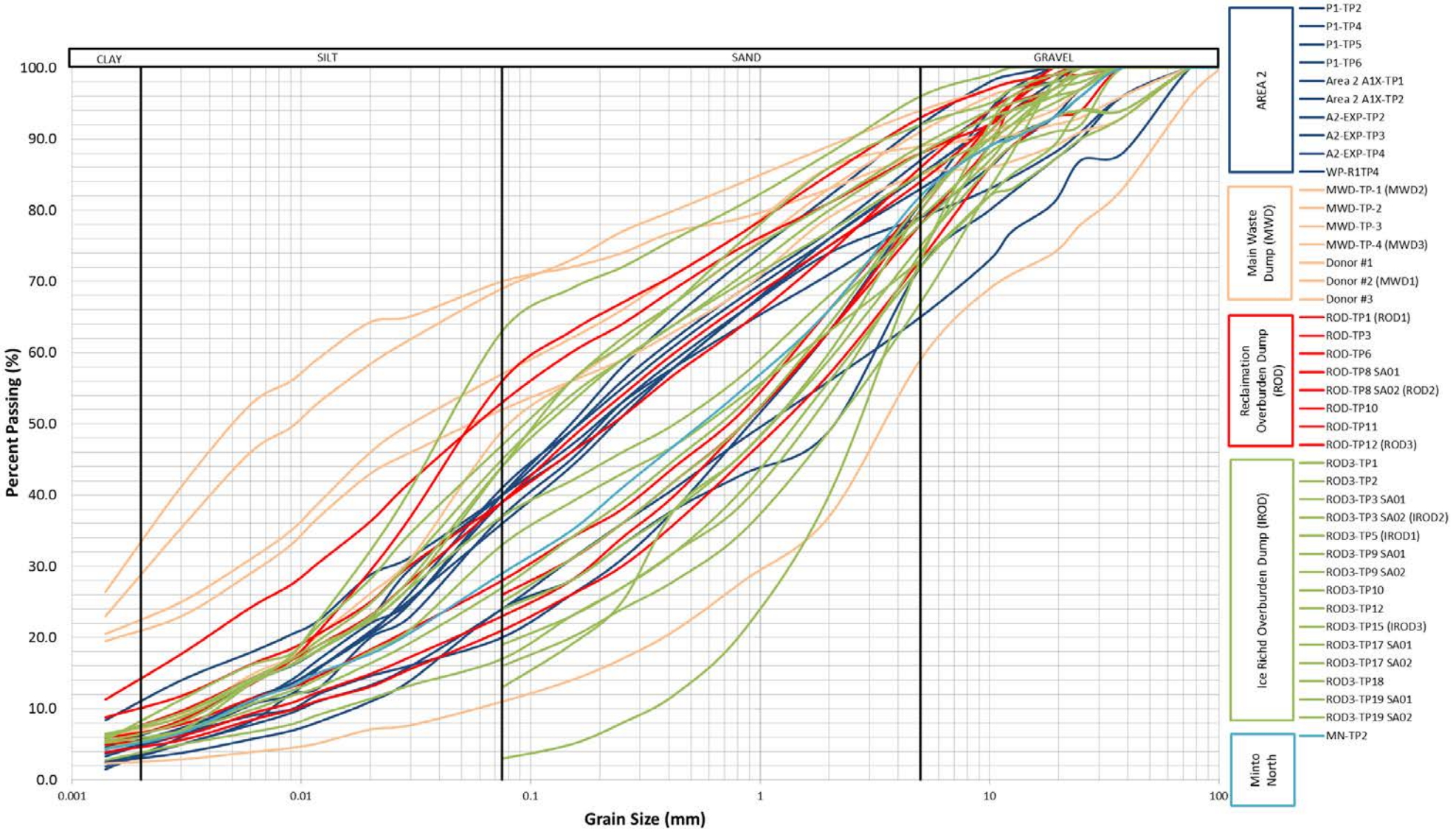
srk consulting	CAPSTONE MINING CORP. MINTO MINE	Scoping Level Cover Assessment
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srk consulting	CAPSTONE MINING CORP. MINTO MINE	Scoping Level Cover Assessment
Job No: 1CM002.007 Filename: Figure 2-14_1CM002.007_Rev05.pptx	Minto Mine	**Climate Wheel**
		Date: July 2013 / Approved: EMR / Figure: **5**

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Scoping Level Cover Assessment

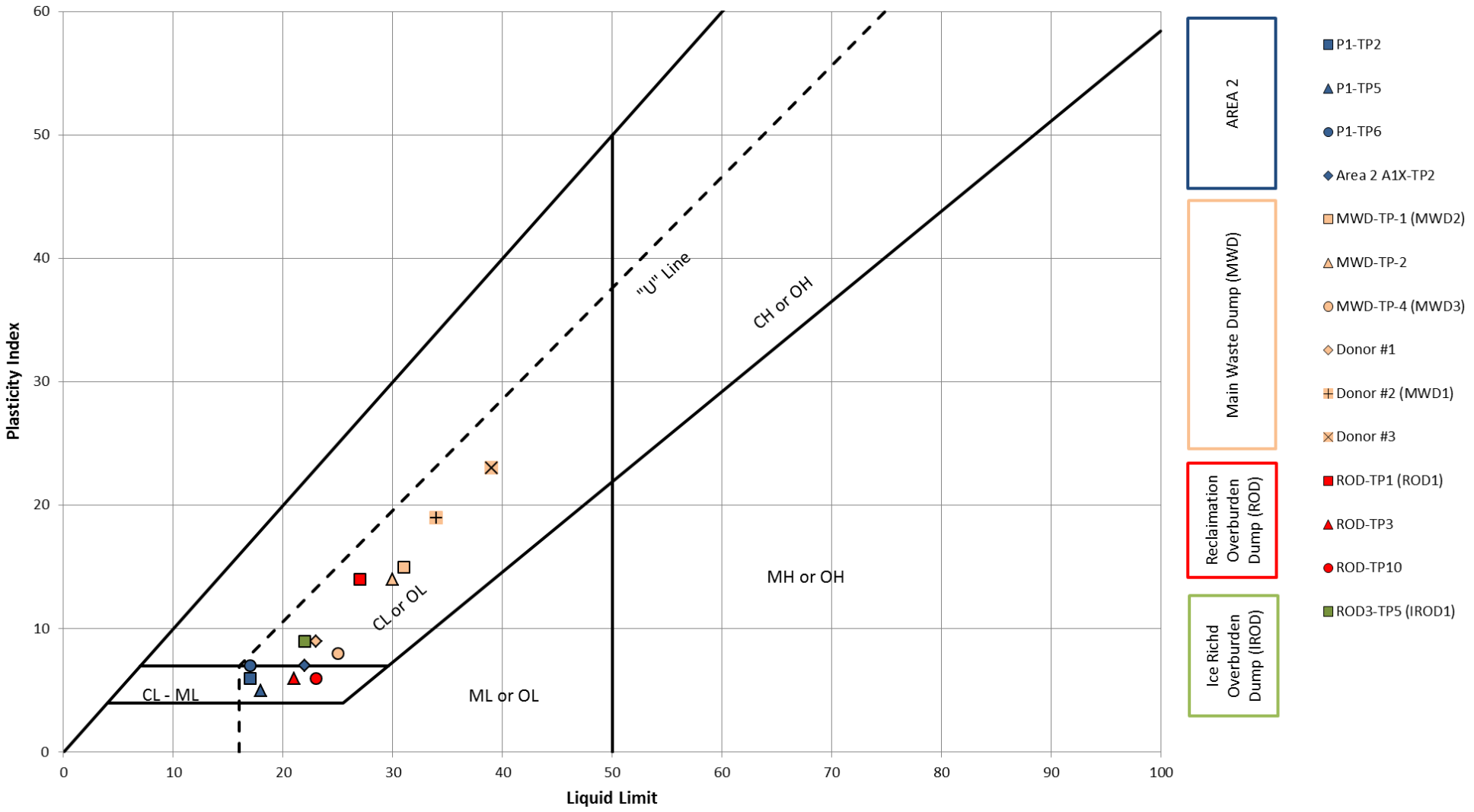
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Minto Mine

Date: July 2013
Approved: EMR
Figure: 6

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Scoping Level Cover Assessment

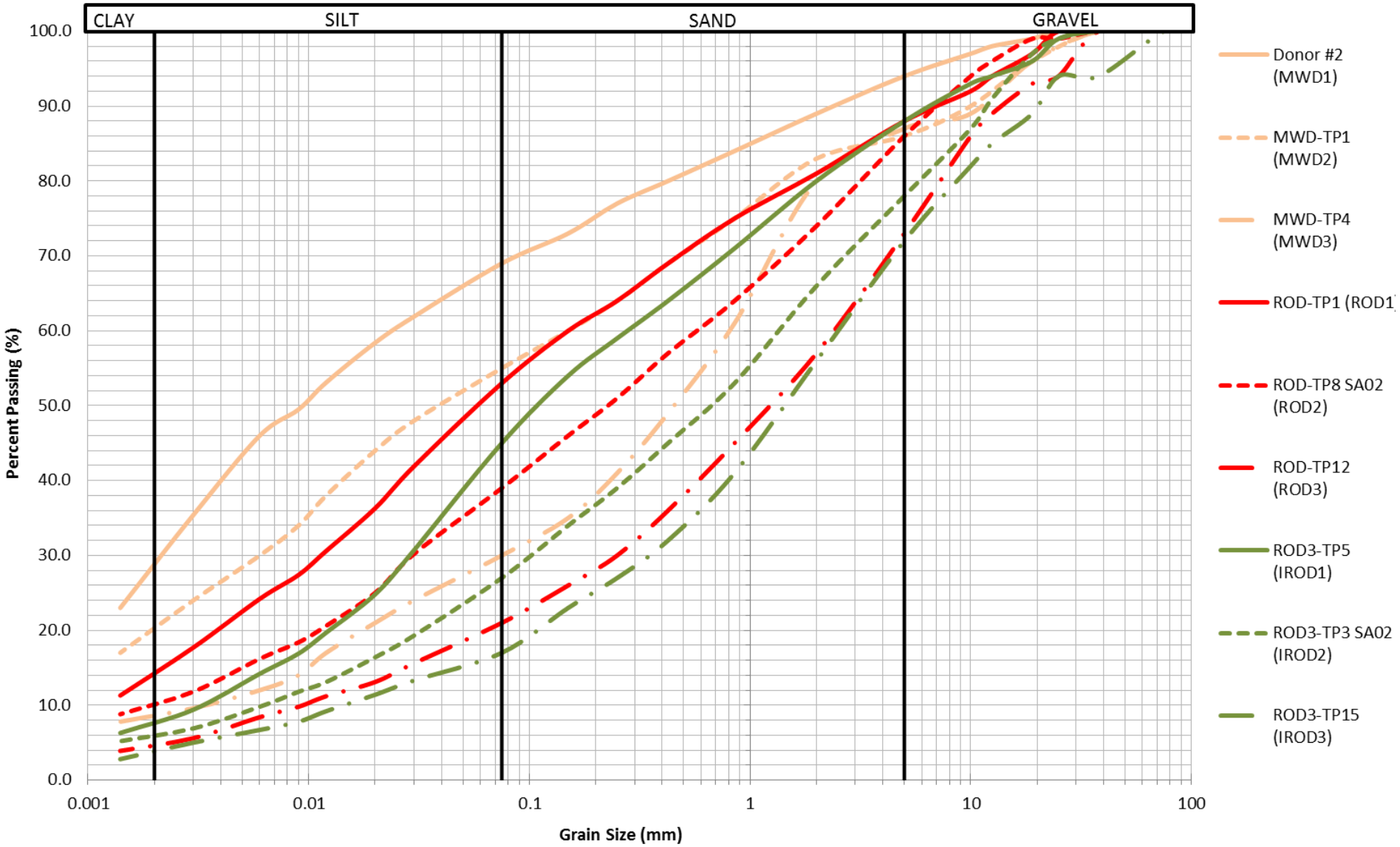
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Filename: Figure 2-14_1CM002.007_Rev05.pptx

Minto Mine

Date: July 2013
Approved: EMR
Figure: 7

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Scoping Level Cover Assessment

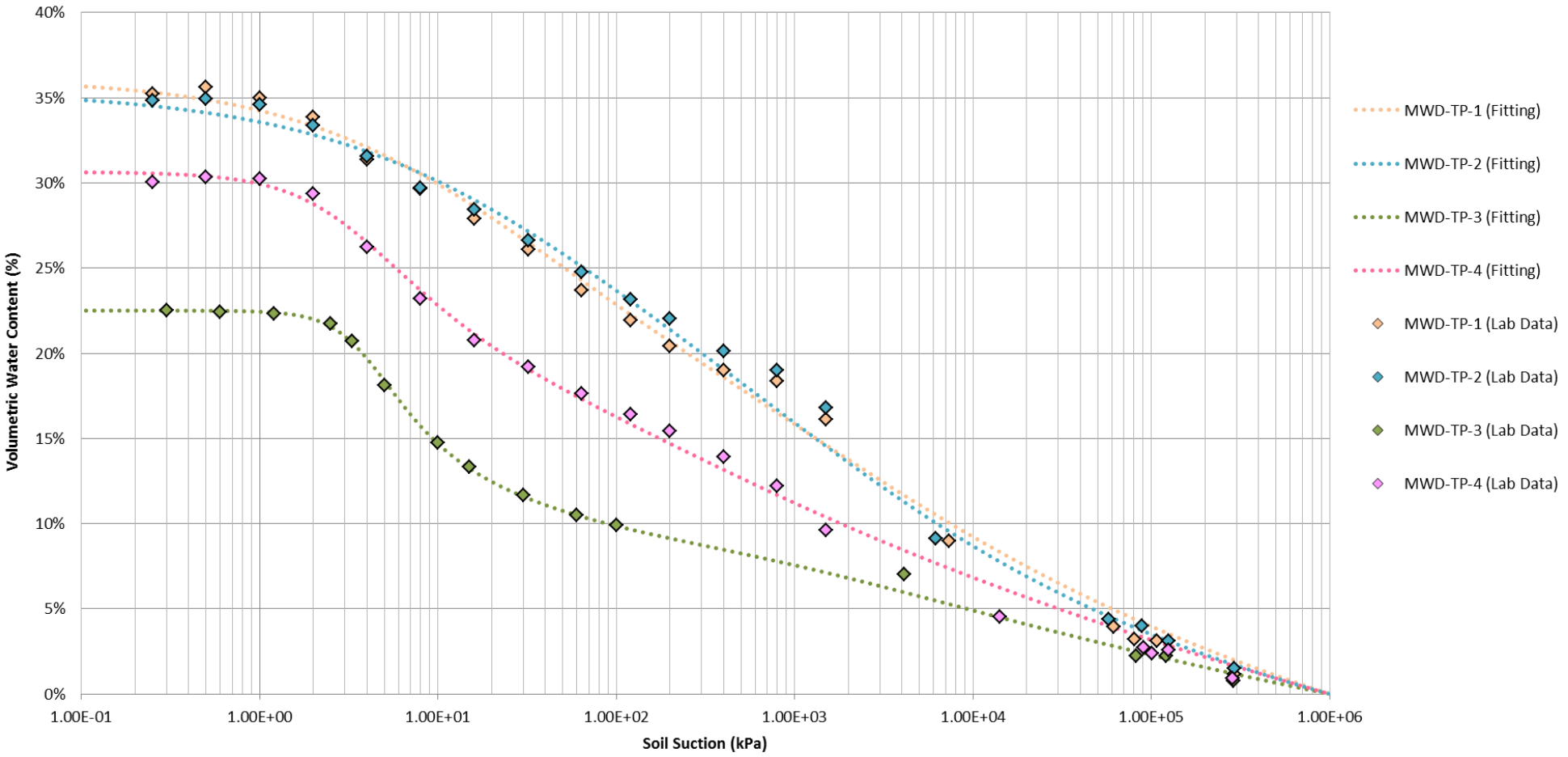
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Minto Mine

Date:	Approved:	Figure:
July 2013	EMR	8

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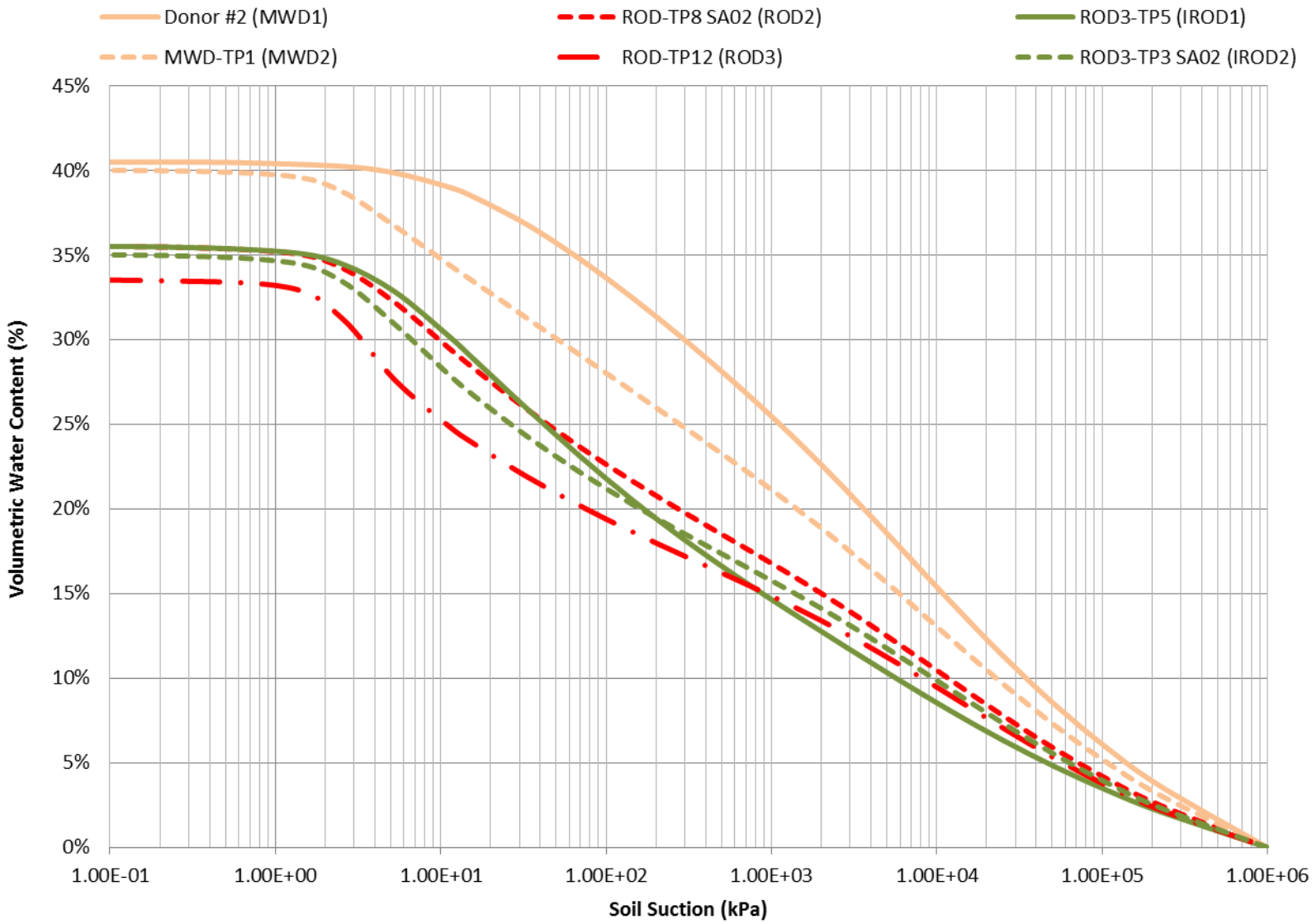
Minto Mine

Scoping Level Cover Assessment

Measured Soil Water Characteristic Curves

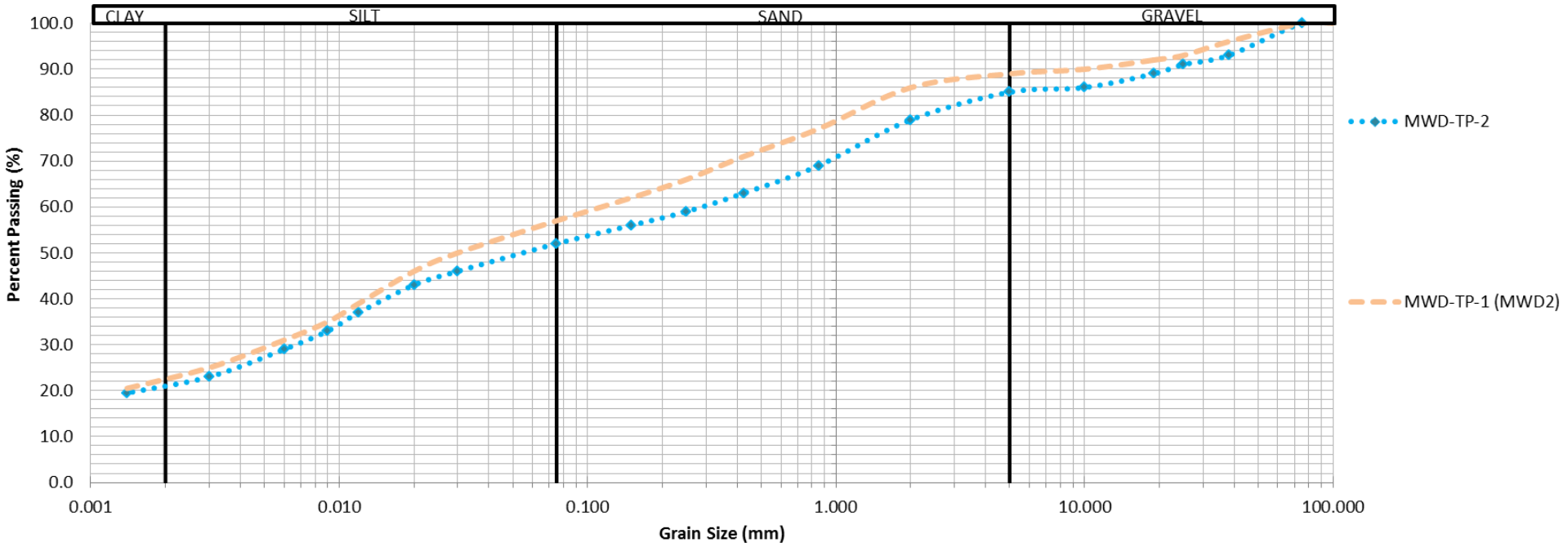
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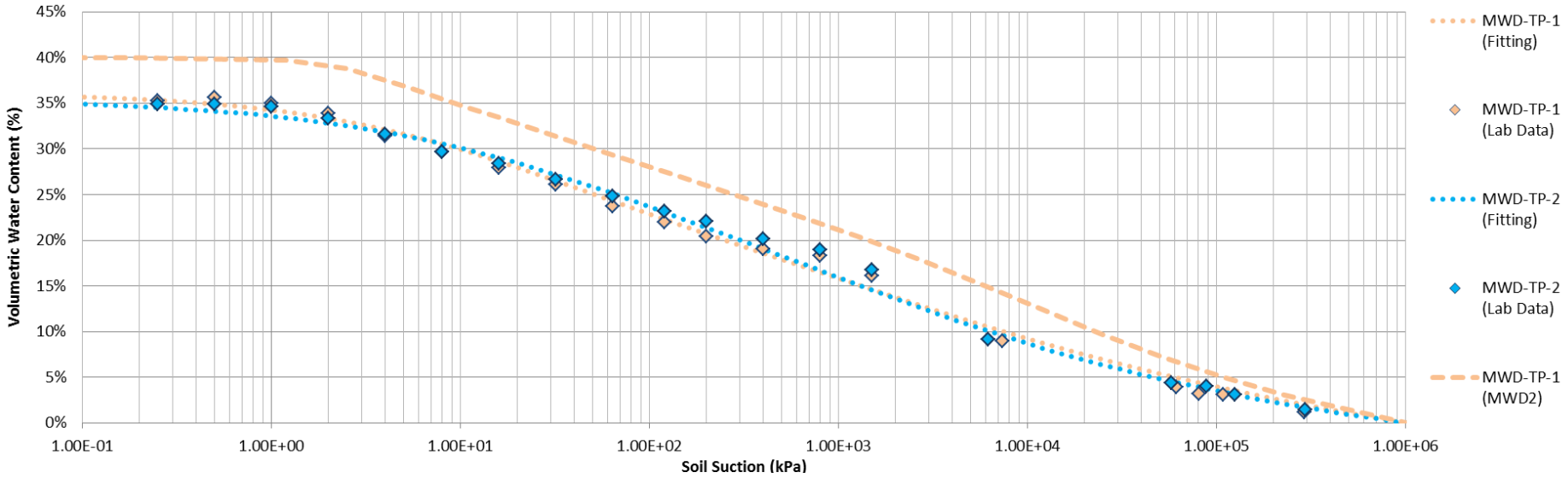
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Job No: 1CM002.007 Filename: Figure 2-14_1CM002.007_Rev05.pptx	Minto Mine	**Estimated Soil Water Characteristic Curves**	
		Date: July 2013	Approved: EMR
		Figure: **10**	

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Scoping Level Cover Assessment

Comparison of Measured and Estimated SWCC's (MWD Material)

Job No: 1CM002.007

Filename: Figure 2-14_1CM002.007_Rev05.pptx

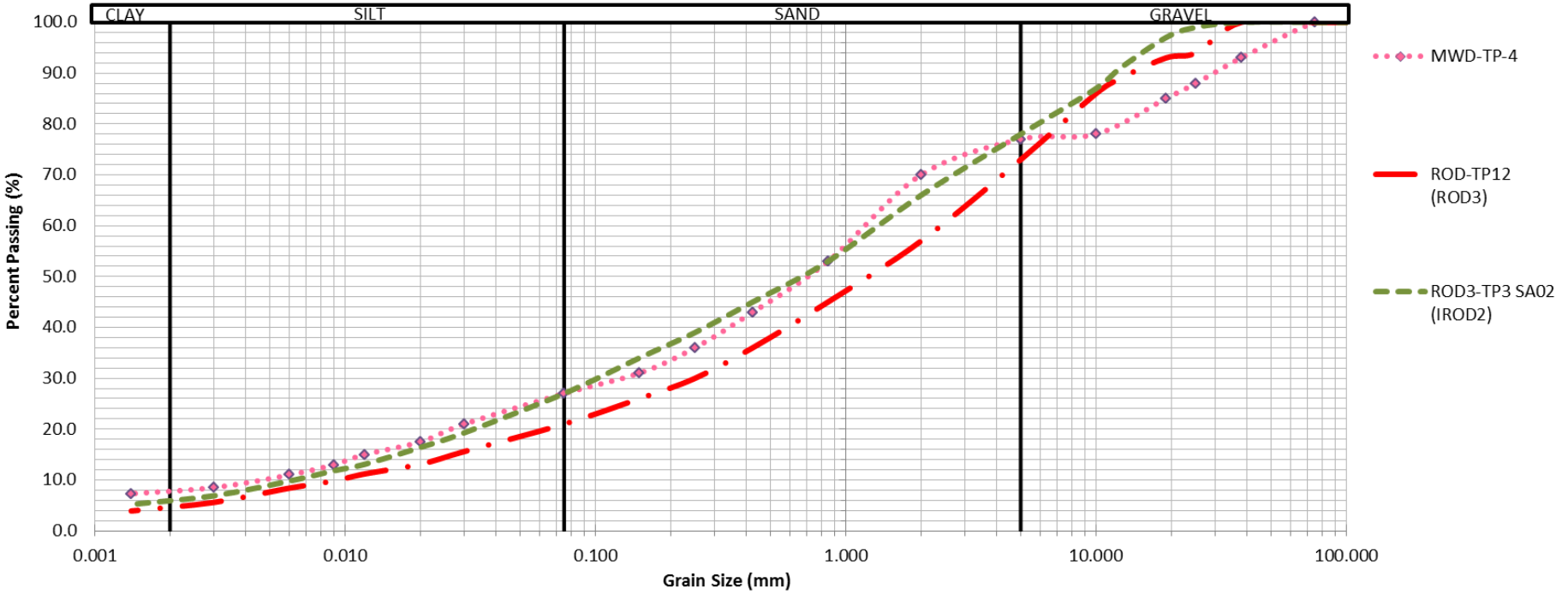
Minto Mine

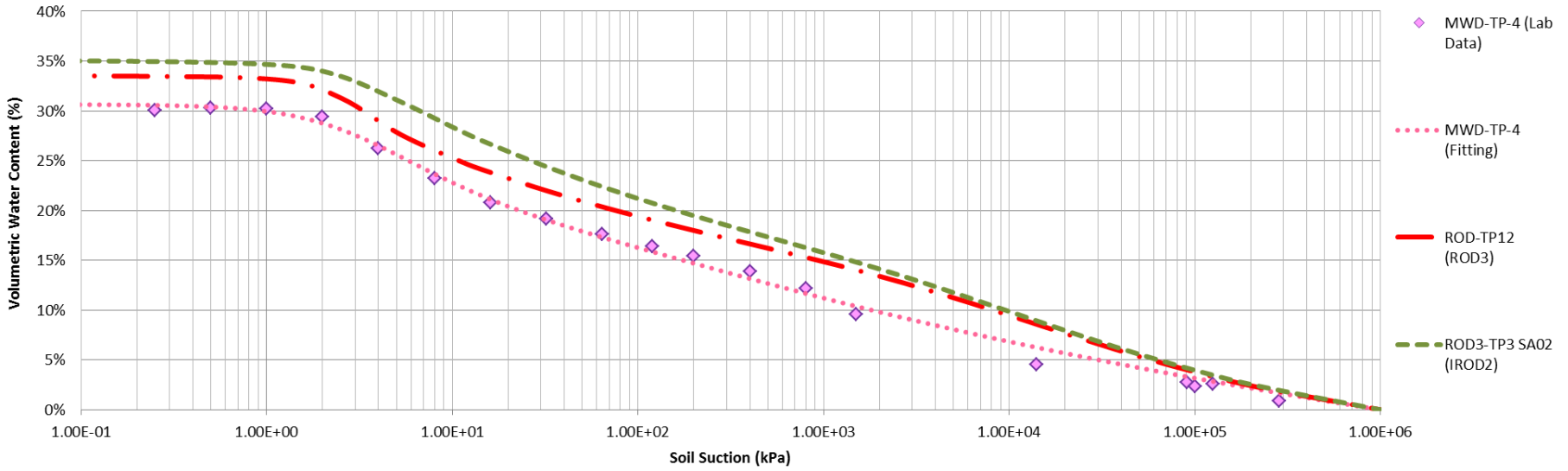
Date: July 2013

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Figure: 11

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Scoping Level Cover Assessment

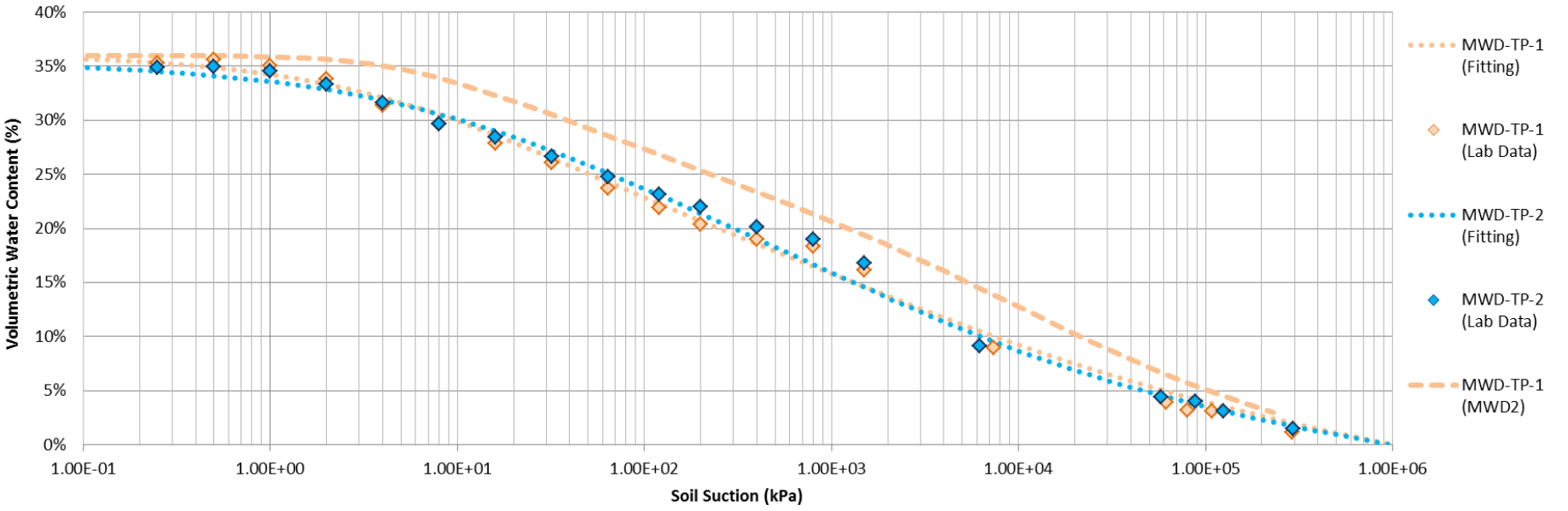
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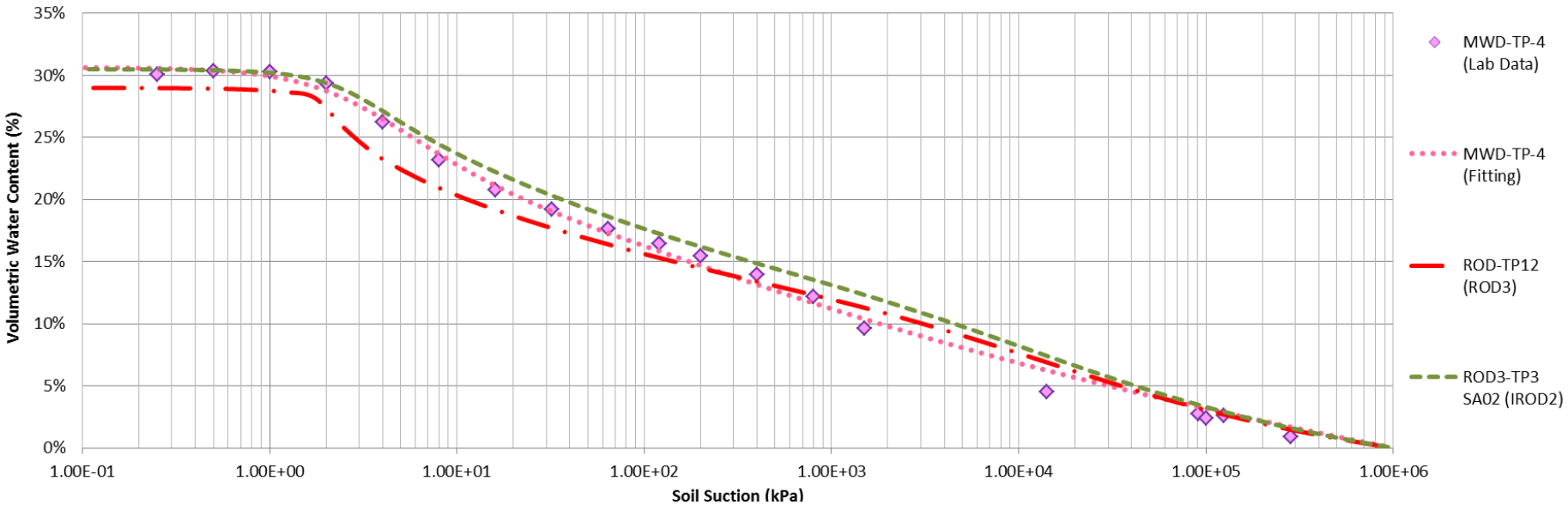
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Minto Mine

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Figure: 12

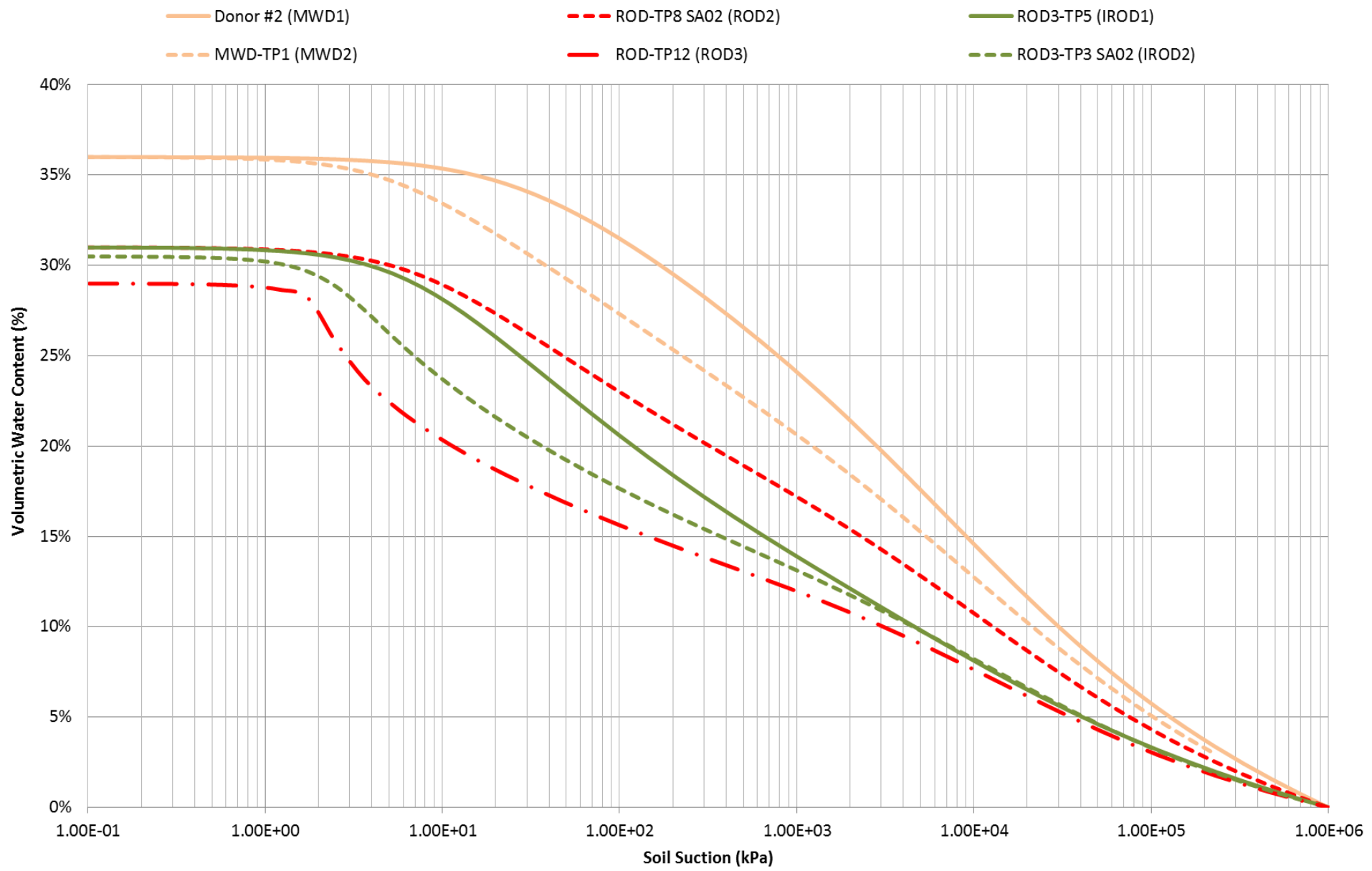
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Job No: 1CM002.007 Filename: Figure 2-14_1CM002.007_Rev05.pptx	Minto Mine	**Calibrated SWCC Estimates Compared to Measured Data**
		Date: July 2013 / Approved: EMR / Figure: **13**

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Scoping Level Cover Assessment

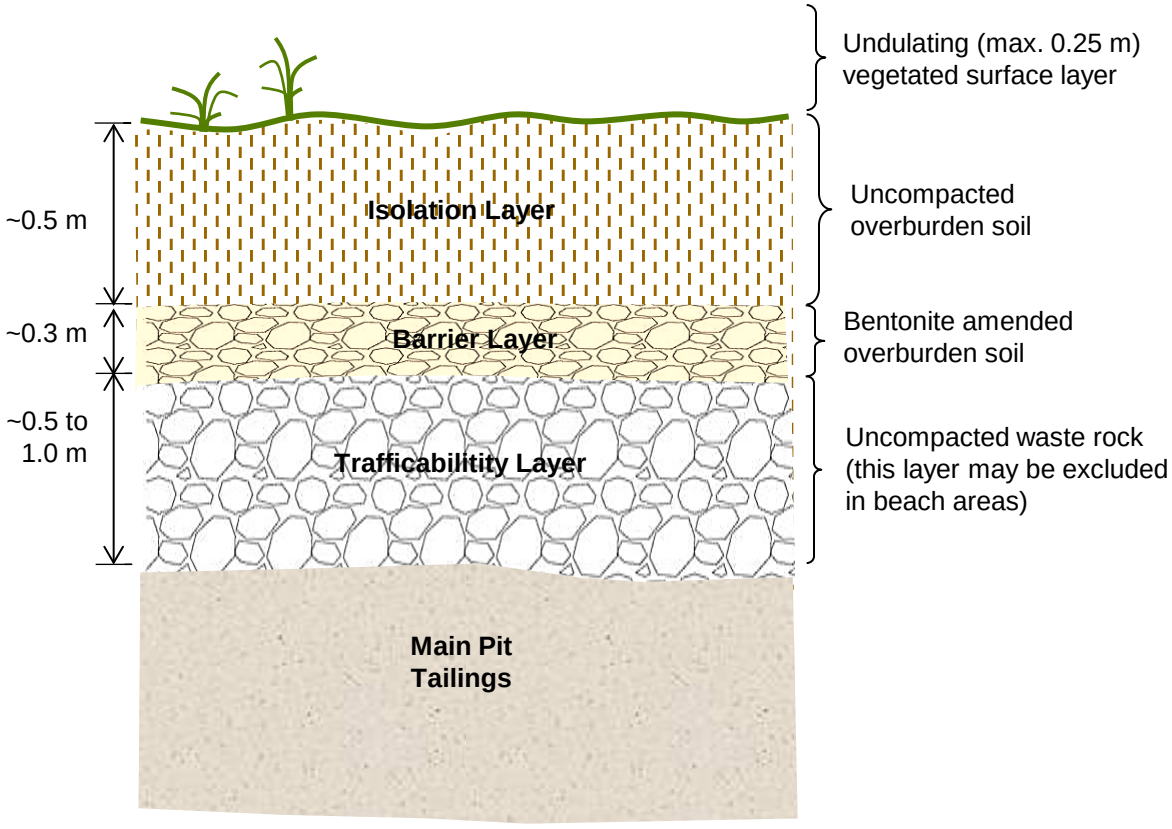
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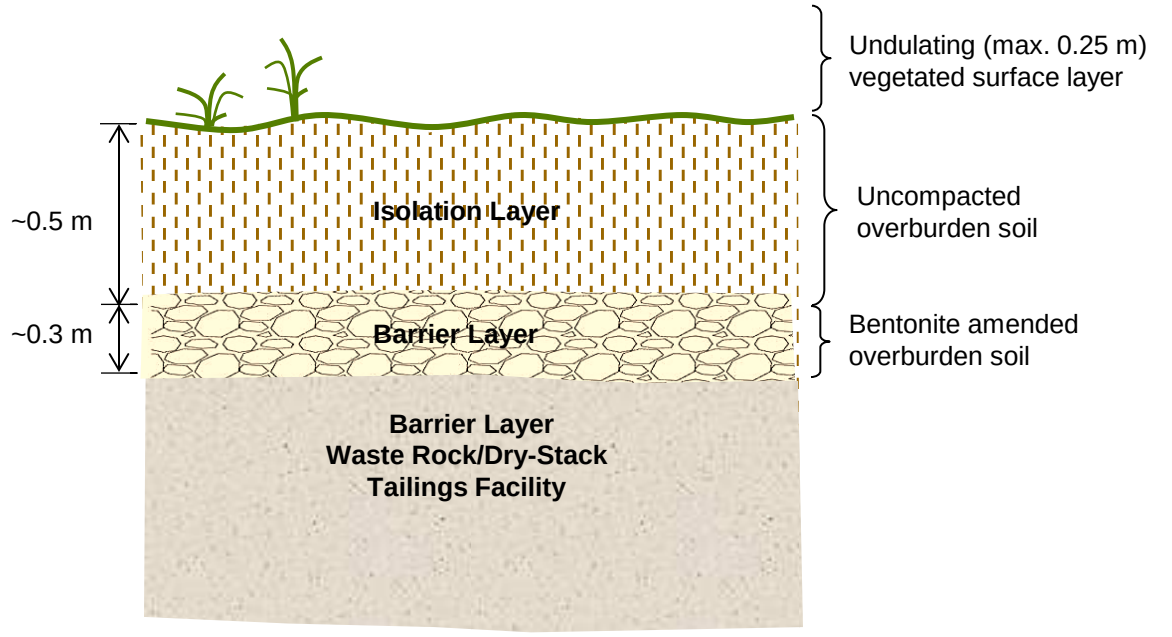
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Minto Mine

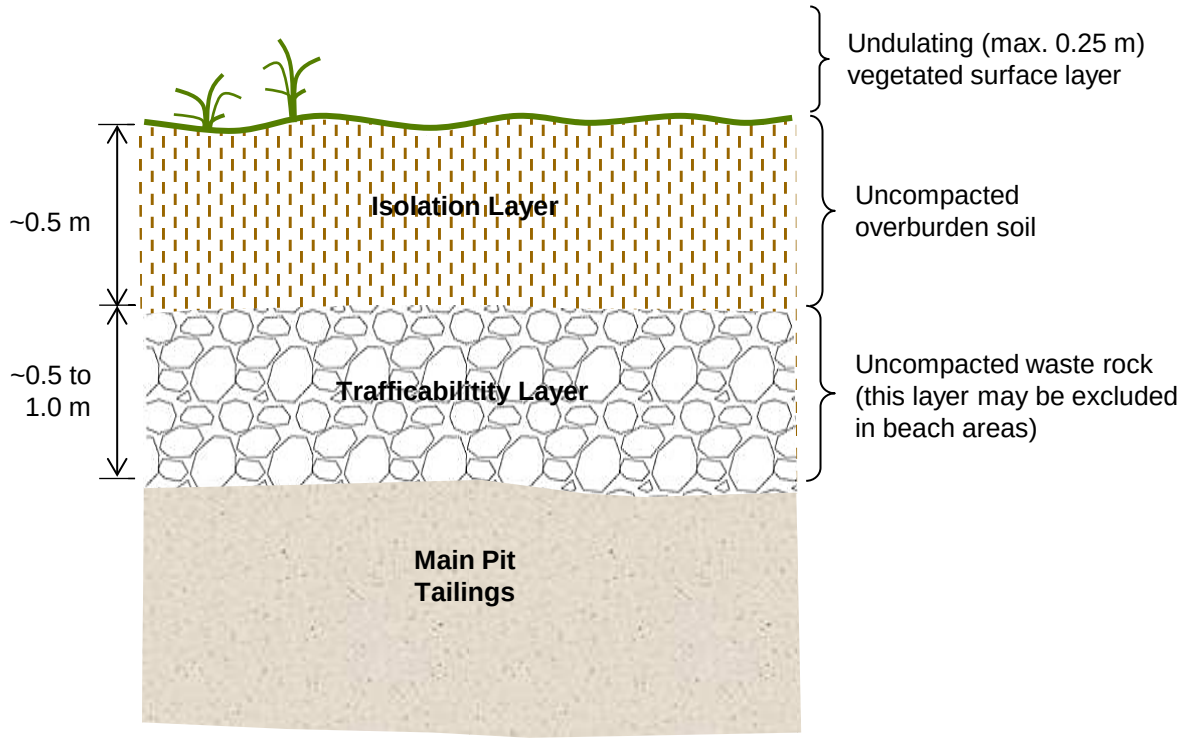
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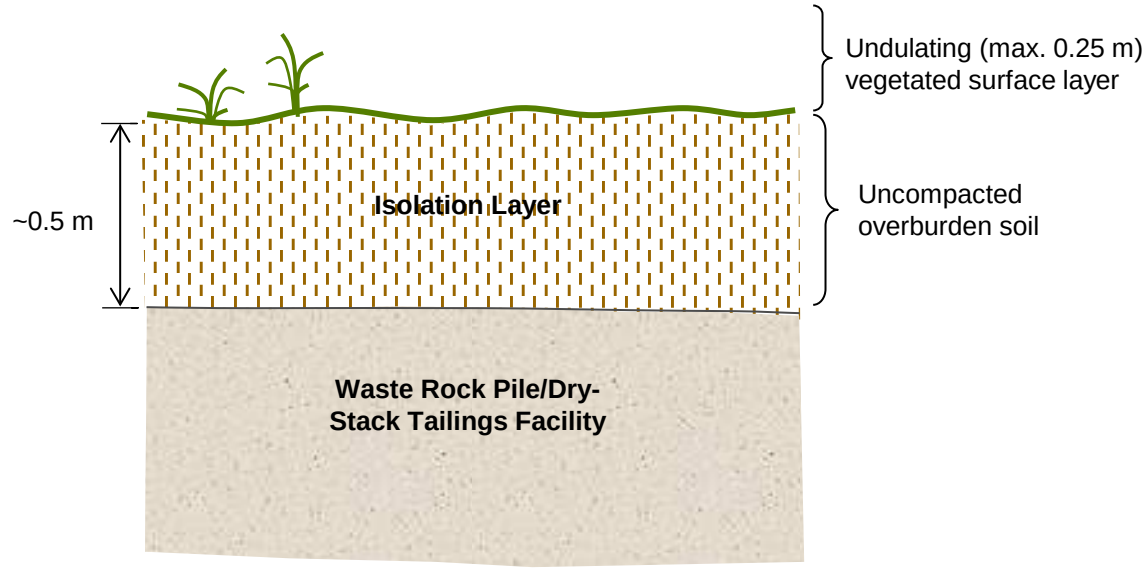
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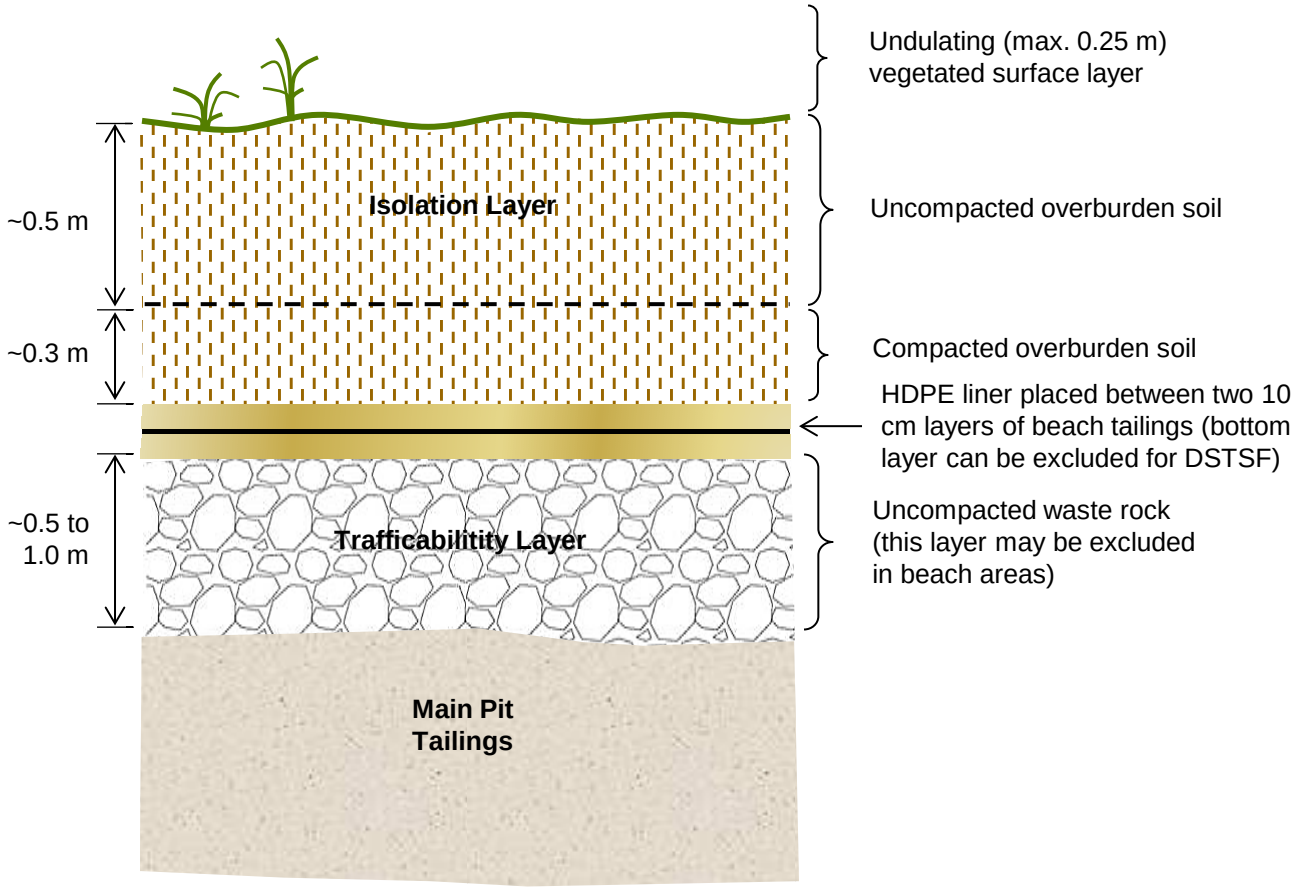


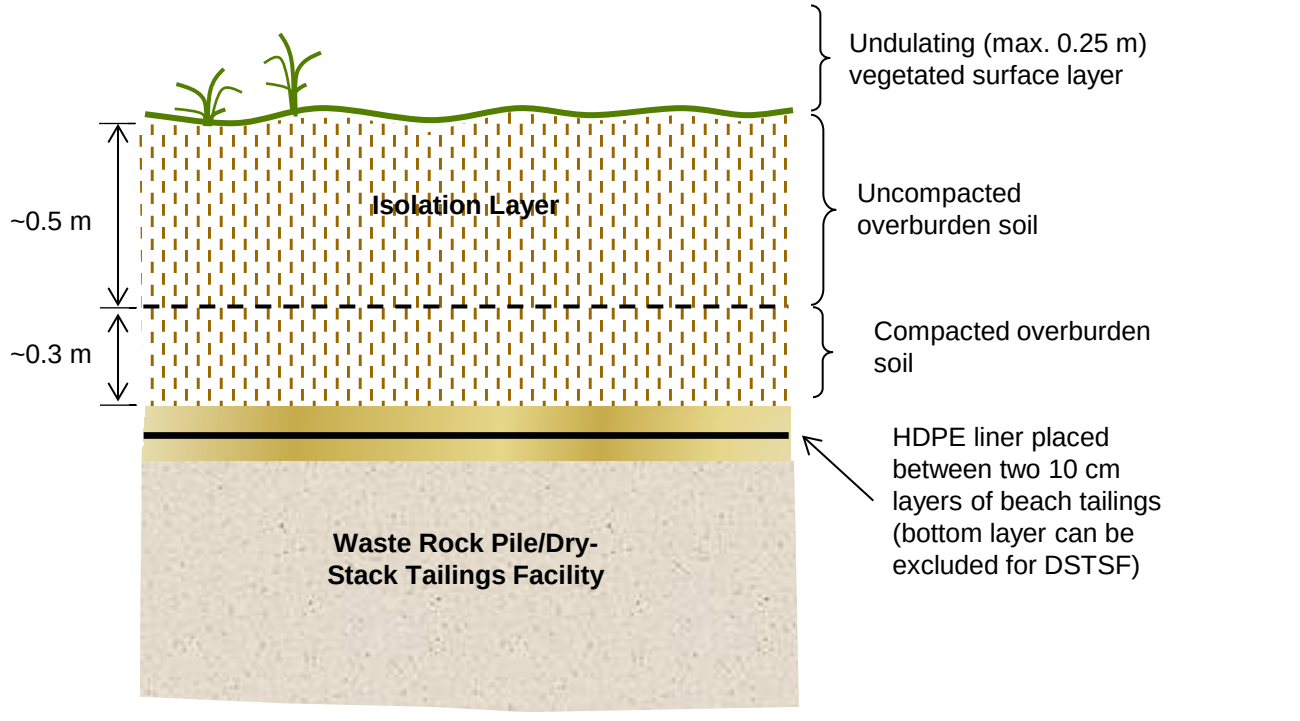
srk consulting	CAPSTONE MINING CORP. MINTO MINE	Scoping Level Cover Assessment		
		Conceptual Low Infiltration Cover		
Job No: 1CM002.007 Filename: Figure 15-17_1CM002.0007_Conceptual Covers	Minto Mine	Date: July 2013	Approved: EMR	Figure: **16**





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		Conceptual Isolation Cover		
Job No: 1CM002.007 Filename: Figure 15-17_1CM002.0007_Conceptual Covers	Minto Mine	Date: July 2013	Approved: EMR	Figure: **15**





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		Conceptual Very Low Infiltration Cover		
Job No: 1CM002.007 Filename: Figure 15-17_1CM002.0007_Conceptual Covers	Minto Mine	Date: July 2013	Approved: EMR	Figure: **17**

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Memo

To:	Ryan Herbert, Minto Explorations Ltd.	**Client:**	Minto Explorations Ltd
From:	Iozsef Miskolczi, PEng, Maritz Rykaart, PhD, PEng	**Project No:**	1CM002.030
Cc:	Dylan MacGregor, PGeo, SRK	**Date:**	October 14, 2015
Subject:	Minto Mine Closure Covers: Results of Numerical Modelling to Bracket Percolation Predictions		

1 Introduction

1.1 General

Preliminary water quality modelling of seepage from Minto Mine waste facilities, as a result of meteoric percolation through them, suggests that under worst-case load predictions the water quality in Minto Creek will not meet the post-closure objectives. Closure covers that control the percolation rate are one of many possible strategies that are being considered to mitigate this situation.

A Scoping Level Cover Assessment was completed for the Minto mine to evaluate what the most suitable closure cover may be (SRK 2013a) given the site specific climatic conditions and availability of candidate cover materials. The scoping study identified several candidate closure cover designs and the approximate range of net percolation that could be associated with each cover concept. Those percolation ranges were based on engineering judgement derived from knowledge of soil covers as well as soil cover performance at other sites, as opposed to measured or predicted (i.e. modelled) rates.

This memorandum presents the results of surface flux boundary modelling that further defines and brackets the likely range of percolation for the proposed soil covers at Minto Mine. These results supplement the existing database of information and provide greater confidence in the use of closure covers as a mitigation strategy.

1.2 Modelling Approach

For the purpose of this study we have adopted the term percolation, as opposed to infiltration, to define the meteoric water that will migrate through the cover and underlying waste, and emerge as seepage at the toe of the facility or report to groundwater. In order to model percolation, a surface flux boundary model must be used, which accounts for the physical processes (e.g. infiltration, evapotranspiration, and runoff) that take place at the cover surface (i.e. at the interface between the cover and the atmosphere). This type of modelling is complex, requires good quality input data, and ideally, long-term in-situ calibration data.

Since no calibration data is available, the modelling approach was to develop a most realistic base case scenario, and then through sensitivity analysis establish how the modelling results will be impacted as parameters change. This process allows for a rigorous assessment of the parameters that drive the cover performance, such that the expected percolation rates can be determined with a high level of confidence.

In order to do this most efficiently, a pseudo- surface flux boundary model (Hydrus 1D (Šimůnek *et al*. 2009)) was used for the modelling. The base case scenario was subsequently verified using a true surface flux boundary model, SVFlux (SoilVision 2009). A true surface flux boundary model calculates actual evaporation using the Modified Penman equation (Wilson 1990), while in the pseudo surface flux boundary model actual evaporation is estimated based on a user defined potential evaporation. Calculating actual evaporation requires a higher level of complexity (i.e. non-linearity) in the numerical simulation which substantially increases the computing time. It was however demonstrated that with proper verification the approach as adopted (i.e. modelling using Hydrus 1D and verification using SVFlux) is reliable and efficient (Rykaart and Noël 2003).

2 Methods

2.1 Conceptual Model and Model Setup

The objective of the modelling is to bracket the likely range of percolation through various single layer soil covers that could be constructed using the candidate soil material that is readily available at Minto Mine. The model assumes that no water table is present within the waste substrate and that any water that percolates through the cover will continue to drain unimpeded through the underlying waste exiting the model at the bottom. In real life however, at the waste dump scale this drainage water will either emerge as toe seepage or report to groundwater.

This conceptual model is represented in the computer models as a 10 m high, one-dimensional column consisting of two layers: a soil cover (of variable thickness) and underlying waste material (waste rock or tailings) respectively. The top surface is flat, however the simulations do not allow for ponding to develop; this corresponds to the physical equivalent of shedding excess water (if it were to occur) as runoff.

In each case the model is run for a continuous period of 20 years in daily time-steps, using a synthesized climatic dataset. This approach results in 20 different annual percolation rates, representing a range of expected outcomes taking into consideration variability of actual climatic conditions. The synthetic climatic data set does not take into consideration climate change.

2.2 Model Input Data

2.2.1 Meteorological Data

Data Source

Daily precipitation and evaporation data is required for modelling. In order to develop a long-term modelling dataset, daily regional meteorological data was obtained from Environment Canada's Pelly Ranch weather station. The Pelly Ranch data, which dates back to 1955, were subsequently modified to better represent Minto site conditions by applying site specific climate

correlation factors that were previously developed based on data from two on-site weather stations (SRK 2012).

Precipitation

The Pelly Ranch meteorological data was parsed for incomplete years and an initial 49-year daily dataset was compiled stretching from 1957 to 2009. This dataset was further purged to remove all years with total precipitation above the 1:50 years wet year (461 mm) and below the 1 in 50 years dry year (218 mm) (SRK 2012). The twenty most recent years from the resulting synthetic dataset were used as model input, from the period between 1986 and 2009, with the years 1992, 1996, 1998 and 2008 excluded. A summary of this annualized dataset is presented in Table 1.

Table 1: Yearly Total Precipitation Data for 20-Year Modelling Period

Model Year	Calendar Year	Yearly Total Precipitation (mm)
1	1986	316
2	1987	362
3	1988	301
4	1989	254
5	1990	372
6	1991	362
7	1993	335
8	1994	293
9	1995	307
10	1997	425
11	1999	417
12	2000	453
13	2001	326
14	2002	335
15	2003	330
16	2004	422
17	2005	367
18	2006	261
19	2007	349
20	2009	349

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.030_Closure_Cover_Modelling\!080_Deliverables\Cover Model Results\020_Tables\[Minto_WeatherInput_1CM002.030_Rev01_IM.xlsx]Precip

To study the effect of higher or lower than average precipitation, two subsets of precipitation data, each 5 years long were compiled. The wetter than average dataset is comprising of years exceeding the 1:5 years wet annual precipitation of 380 mm (SRK 2012), as presented in Table 2. The drier than average dataset includes years in which total precipitation was below 275 mm, or 1:5 years dry (SRK 2012) as shown in Table 3.

Table 2: Wetter Than Average Precipitation Years

Model Year	Calendar Year	Yearly Total Precipitation (mm)
1	1964	388
2	1967	408
3	1981	380
4	1985	382
5	2000	453

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.030_Closure_Cover_Modelling\!080_Deliverables\Cover Model Results\020_Tables\[Minto_WeatherInput_1CM002.030_Rev01_IM.xlsx]Precip

Table 3: Dryer Than Average Precipitation Years

Model Year	Calendar Year	Yearly Total Precipitation (mm)
1	1969	260
2	1971	269
3	1975	270
4	1980	257
5	2006	261

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.030_Closure_Cover_Modelling\!080_Deliverables\Cover Model Results\020_Tables\[Minto_WeatherInput_1CM002.030_Rev01_IM.xlsx]Precip

Evaporation

Potential evaporation (PE) was estimated at about 400 mm/year by scaling the lake evaporation data from Pelly Ranch to match the Minto Mine site conditions (SRK 2013a). The monthly total values of PE, shown in Table 4, were obtained by scaling monthly PE values calculated using the Penman-Monteith method (FAO 1998) to match the yearly total of 400 mm.

Table 4: Monthly Distribution of Potential Evaporation

Month	Monthly PE (mm)	Days per Month	Daily PE (mm/day)
Jan	0	31	0.00
Feb	0	28.25	0.00
Mar	0	31	0.00
Apr	0	30	0.00
May	85	31	2.75
Jun	98	30	3.28
Jul	92	31	2.96
Aug	82	31	2.63
Sep	38	30	1.28
Oct	5	31	0.16
Nov	0	30	0.00
Dec	0	31	0.00

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.030_Closure_Cover_Modelling\!080_Deliverables\Cover Model Results\020_Tables\[Minto_WeatherInput_1CM002.030_Rev01_IM.xlsx]Potential Evap

Evapotranspiration

Bare ground evaporation accounts for the water loss at the soil-atmosphere boundary, i.e. the soil surface. Plant transpiration on the other hand is responsible for removal of water from the near-surface of the soil to the full depth of the rooting zone. During the growth season plant transpiration and bare ground evaporation occur simultaneously and the coupled water loss is termed evapotranspiration (Hillel 1980). Bare ground evaporation can however occur outside of the growth season, as long as water at the soil surface is available in liquid form.

The site experiences a short growth season. Based on temperature records from the Pelly Ranch station the average growth season over the most recent 17 years (1998 to 2014) is between May 5 and September 8 (Access 2014). The beginning of the season was considered the fifth consecutive day with mean air temperature above +5°C whereas the end was the first day with killing frost (below -2.2°C). Potential bare ground evaporation (PE) however, which is based on the energy balance governed by the local climate, has a wider range extending between May 1 and October 31. The model assumes that evapotranspiration is possible within this period only, whereas it is considered to be zero between November 1 and April 30.

Snowmelt and Sublimation

To recognize the effect of freezing conditions during winter, snow precipitation was retained and accumulated between October 1 and April 17 annually as snow water equivalent (SWE). Snowmelt was assumed to start on April 17 with freshet lasting two weeks from April 17 to April 30. The entire accumulated SWE was then released as equal daily amounts of meltwater over these two weeks, in addition to any observed precipitation. The ground was assumed to be fully thawed during this period; this is conservative, as in reality thawing would only be starting at this time.

Snow sublimation in the Yukon can lead to considerable snow cover loss. A rigorous study of sublimation performed at a research station near Whitehorse (Pomeroy *et al.* 1999) observed sublimation ranging between 28 and 45 mm per season. Since winter climate conditions influencing the magnitude of sublimation are comparable between Minto and the study site, it was considered that 45 mm of sublimation are representative of Minto site conditions. For modelling purposes, the April 17 SWE was reduced by 45 mm as an allowance for sublimation.

2.2.2 Material Properties

Field investigations completed in 2012 characterized all candidate cover material sources. Tailings samples were obtained from the Dry Stack Tailings Storage Facility and analysed as part of a field investigation conducted in 2013 (SRK 2013b). Particle size distribution (PSD) curves encompassing the full range of candidate cover materials on site are reproduced in Figure 1. Advanced hydraulic testing, including Soil Water Characteristic Curves (SWCC), was performed on representative samples.

Table 3 shows a summary of the candidate cover material properties used in the Hydrus 1D model, while Figure 2 and Figure 3 present the PSD and SWCC curves used in modelling.

A literature search was performed to gather properties for the waste rock to be used in modelling. The material considered to be most appropriate for representing the Minto waste rock was a

waste rock sample from Greens Creek mine (Hopp *et al.* 2011) due to the igneous (hard rock) nature of both the Greens Creek and Minto geology. For the sensitivity analysis, finer waste rock properties were used from SRK's database, while the coarser waste rock was simulated using van Genuchten curve fitting parameters (van Genuchten 1980). The water retention properties for tailings were predicted using the Rosetta model (Schaap *et al*. 2001), based on a Minto sample PSD with a bulk density of 1.9 T/m^3. No SWCC curves are available for the tailings and coarse waste rock materials, and therefore modelling was done using the unsaturated curve fitting parameters listed in Table 5.

Table 5: Summary of Material Properties Used in Modelling

Materials ID	Material Type	θr	θs	α (1/m)	n	K_{sat} (m/day)	l
MWDTP4 [a]	Cover	0	0.303	2.67	1.20	0.164	0.5
MWDTP2 [a]	Cover	0	0.360	1.09	1.19	0.025	0.5
MWDTP3 [a]	Cover	0	0.238	5.50	1.22	3.715	0.5
Greens Creek WR [b]	Waste Rock	0.012	0.410	5.43	2.03	1.728	0.5
Finer WR [c]	Waste Rock	0	0.361	32.65	1.22	4.790	0.5
Coarser WR [d]	Waste Rock	0.020	0.450	2.00	2.50	5.000	0.5
Tailings [e]	Tailings	0.027	0.272	6.31	1.22	0.081	0.5

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.030_Closure_Cover_Modelling\Task_300-CoverModel\Hydrus1D\ Minto_CoverHydrusModel_Summary_Rev06_KK_IM.xlsx

Notes:

[a] : Based on RETC model using van Genuchten – Mualem method; K_sat based on laboratory result.
[b] : Hopp, L. et al. 2011.
[c] : Based on RETC model using van Genuchten – Mualem method; K_sat based on laboratory result.
[d] : Theoretical van Genuchten curve fitting values representative of a generic waste rock.
[e] : Estimated values based on particle size distribution and density of tailings on site.

2.2.3 Boundary Conditions

The top boundary condition of the one-dimensional column is a time-dependent (daily time step) atmospheric boundary (precipitation and evaporation). The bottom boundary condition is a unit gradient to simulate unsaturated gravitational flow exiting the model. Being a one-dimensional model, there are no side boundaries.

2.2.4 Initial Conditions

Initial conditions were expressed in terms of gravimetric moisture content, set at 7% for waste rock, 18% for tailings, and 11% for the soil cover materials, respectively. These moisture contents represent the field moisture content of the soil samples collected during the field programs in 2012 (SRK 2013c) and 2013 (SRK 2013b), representing reasonable initial conditions for the model.

2.2.5 Vegetation

Vegetation on site is dominated by mixed Trembling Aspen and Lodgepole Pine forests at various stages of succession following the relatively frequent forest fires. Willow species dominate the shrub covered area, while being ubiquitous in the understory of the forested areas (Access 2013).

No specific information of actual plant transpiration rates was available for model calibration. Therefore the model made use of a conservative pasture-type vegetation, which is one of the Hydrus 1D built-in vegetation functions. A constant rooting depth of 10 cm was applied.

2.3 Scenarios Evaluated

2.3.1 Base Case Scenario

The base case scenario assumes a simple 0.5 m thick soil cover overlying waste rock. The cover material is represented by test data from sample MWDTP4 (SRK 2013c) and the waste rock is represented by the Greens Creek sample. The surface is assumed to have no vegetation. Normal precipitation is applied as a 20-year long daily sequence, with snow precipitation retained during the winter months (simulating snow accumulation on frozen ground), before being reduced by an amount of 45 mm per year to account for sublimation losses. The remaining snow water equivalent is subsequently release to simulate a freshet lasting 14 days starting April 17 each year. PE equal to 400 mm annually is distributed monthly in equal daily increments (see Table 2).

The base case scenario was simulated using both Hydrus and SVFlux. Since SVFlux is a more rigorous model, it allows for a confirmation on whether the Hydrus model result is reasonable.

2.3.2 Sensitivity Analysis Scenarios

The objective of the sensitivity analysis was to evaluate which parameters have the largest influence on the percolation predictions. To that end the following sensitivity runs were completed:

- A delayed freshet (starting two weeks later, on May 1);
- Finer and coarser cover material;
- Finer and coarser underlying waste rock material;
- No sublimation and increased sublimation;
- Increased cover thickness;
- Upset climatic condition; and
- Presence of vegetation.

In addition, to allow for benchmarking of the results, simulations were also completed with no cover. Finally, simulations were completed to determine the percolation rates through these covers if they were placed over tailings.

3 Modelling Results and Discussion

3.1 Results Summary

Table 6 presents a summary of the complete results for all runs, including the sensitivity analyses. Figure 4 presents the results as a chart showing the range of yearly net percolation as percent of total precipitation for that year. The boxed number represents the arithmetic average

of the yearly net percolations over the 20-year simulation, and as such does not represent any particular modelled year.

Table 6: Summary of Modelling Results

Case #	Scenario	Net Percolation (% Annual Precip.)			Total Precip. (mm)	Total Net Perc. (mm)	Total Evap. (mm)	Total Run-off (mm)
		Max.	Min.	Avg.				
1	Base Case	43%	6%	23%	6,038	1,429	4,629	0
2	Late Freshet (May 1)	35%	<1%	14%	6,038	934	5,126	0
3	Finer Cover Material	40%	2%	21%	6,038	1,305	4,717	0
4	Coarser Cover Material	44%	5%	23%	6,038	1,447	4,673	0
5	Finer Waste Rock	46%	9%	23%	6,038	1,428	4,626	0
6	Coarser Waste Rock	49%	8%	28%	6,038	1,749	4,316	0
7	1 in 5 years wet	29%	17%	23%	7,065	1,612	5,468	0
8	1 in 5 years dry	26%	8%	14%	4,607	682	3,949	0
9	Thicker Cover (1 m)	38%	3%	19%	6,038	1,179	4,824	0
10	Thicker Cover (2 m)	36%	3%	18%	6,038	1,134	4,904	0
11	Tailings WR traffic layer under cover	38%	6%	21%	6,038	1,299	4,750	0
12	WR only; No Cover	65%	34%	45%	6,038	2,742	3,291	0
13	Tailings with WR; No Cover	61%	31%	41%	6,038	2,500	3,541	0.3
14	Tailings only; No Cover	60%	26%	39%	6,038	2,383	3,665	0
15	BaseCase; No Sublimation	54%	13%	33%	6,933	2,323	4,651	0
16	Vegetation	40%	4%	21%	6,038	1,298	4,719	0
17	Base case (80 mm sublimation)	30%	1%	14%	5,332	759	4,579	0
18	SV Flux check	44%	5%	19%	6,044	1,228	4,821	0
19	SV Flux check; No Sublimation	54%	11%	29%	6,940	2,090	4,851	0

Source: \\van-svr0\Projects\01_SITES\Minto\1CM002.030_Closure_Cover_Modelling\!080_Deliverables\Cover Model Results\020_Tables\[Minto_CoverHydrus Model_Summary_Rev13_IM.xlsx]Summary

In all cases the results are predicted as a range around an average value as opposed to a single percolation value. This is intentional, as surface flux boundary modelling cannot yield absolute results. The percolation outcome is a function of the complex interaction between antecedent meteoric events and pre-existing soil moisture conditions, which naturally is highly variable. Similar precipitation events can yield vastly different percolation results depending on preceding soil moisture conditions. Therefore the appropriate use of modelling is to bracket the likely range of results, taking into consideration those elements that drive the outcome.

Another way to look at the results is to consider the normal distribution of the range. For example, Figure 5 presents the 20 years of precipitation data, indicating the variability which ultimately resulted in the Base Case model calculating a range of percolation between 6% and 43% with the average of 23%.

Arranging the predicted range into a histogram (Figure 6) reveals that eight out of 20 years the percolation is likely to be in the near-average range, followed by six out of twenty years

moderately exceeding the average. This compares to total precipitation of about 300 and 350 mm respectively, thus representing a rough correlation between the precipitation and net percolation. The correlation however is reliable only for the modelled results and extrapolation outside of the modelled range may not be valid. This is due to the fact that the relationship between precipitation and percolation is not linear, as described earlier in this section.

Using some basic statistical tools, the same Base Case results can be arranged into a normal probability distribution, as shown in Figure 7. This graphic shows that, for example, 90% of the time percolation will be less than 35% of the yearly precipitation; conversely one in ten years can be expected to have net percolation higher than 35%.

3.2 Model Verification

The Base Case (Case #1) model, which uses the Hydrus 1D code yields an average percolation of 23%, with an overall range between 6 and 43%. The equivalent scenario, using the more rigorous SVFlux code (Case #18), yields and average percolation of 19%, with an overall range of 5 to 44%. A second verification check was made for Case #15 which is the Base Case model, but without sublimation. In this case the average percolation is 33%, while for Case #19 (the equivalent SVFlux run) it is 29%. For Case #15 the overall percolation range is between 13 and 54%, while for Case #19 it is between 11 and 54%.

In all instances the overall range of percolation results are near identical, and most certainly within the accuracy range of surface flux boundary modelling. The average percolation is consistently 4% lower for the SVFlux model runs. This difference is ascribed to the fact that for the SVFlux model runs the evaporation is slightly higher resulting in lower percolation. As described earlier, SVFlux is a true surface flux boundary model that calculates actual evaporation for each time step using the Modified Penman equation, and as a result, this higher evaporation is expected. The small difference of 4% is however within the level of accuracy of this type of modelling, and therefore the Hydrus 1D modelling is entirely suitable. The fact that the Hydrus 1D model reports a slightly higher percolation rate suggests a level of conservatism which is entirely appropriate.

3.3 Base Case (0.5 m Soil Cover over Waste Rock)

A 0.5 m soil cover over waste rock reduces percolation by almost half to 23%, and similarly the range drops to between 6 and 43%. This demonstrates that even a nominal soil cover results in a considerable reduction in percolation, and therefore is potentially beneficial as a mitigation strategy.

3.4 Uncovered Waste Rock and Tailings

The average percolation for the uncovered waste rock (Case #12) and tailings (Case #14) are 45 and 39% respectively. The range for uncovered waste rock is 34 to 65%, while for the uncovered tailings its 26 to 60%. Tailings percolation is less due to the finer grained nature of the material, which retains more moisture near surface making some of that moisture subsequently available for evaporation.

3.5 Simple Soil Cover Over Tailings

Although it would be possible to apply a simple 0.5 m thick soil cover over the Minto Dry-Stack Tailings Facility, this would not be possible or practical for hydraulically placed tailings in the Main Pit. To ensure trafficability, a 1 m thick layer of waste rock would likely first be required. The effect that has on the modelled percolation rate is illustrated by Case #13, which increases the average percolation to 41% compared to 39% for the uncovered tailings. In both these cases the overall percolation range is similar as illustrated in Figure 4. This result is plausible, as the waste rock is coarser than the tailings allowing for higher infiltration and less evaporation; however, with the finer tailings only 1 m below the surface, some moisture remains more available during wetter periods, which means the average percolation is slightly less than the uncovered waste rock.

Covering the trafficking layer with a 0.5 m simple soil cover (Case #11) has the net effect of reducing the average percolation to 21% and the range to between 6 and 38% which is similar to the performance of the Base Case (Case #1).

3.6 Effect of Cover Material Composition

As illustrated in Section 2.2.2, the available candidate cover materials have a significant range in material properties. Modelling the finer and coarser ends of the spectrum (Case #3 and #4 respectively) suggest that this variability does not materially influence the outcome, since the variability in average percolation between the three cover materials is only 2%. This confirms that for the materials characterized to date there is no preferred material type with which lower percolating covers can be constructed.

3.7 Effect of Cover Thickness

Increasing the cover thickness from 0.5 m to 1 m (Case #9) has the effect of reducing net percolation by about 4%, and the overall range in percolation becomes slightly smaller. Further increasing the cover thickness to 2 m (Case #10) results in a negligible additional improvement, which is arguably less than the model accuracy. This outcome is consistent with the cover material properties, in that the capillarity of the cover material is around 1 m, which means that moisture that passes beyond this limit is unlikely to be released via evaporation.

3.8 Effect of Waste Rock Material Composition

Model simulations with a finer (Case #5) and a coarser (Case #6) waste rock material demonstrate a reasonable sensitivity when measured against the Base Case cover. For the finer waste rock there is no effect on the average percolation rate; however, the range shifts towards the higher end as illustrated in Figure 4. When modelling a coarser waste rock, the average net percolation increases to 28%, and the overall range shifts up (similar to the finer waste rock) to between 12 and 49%. This result is consistent with the observations in Section 3.7 above, suggesting that a cover thickness of 0.5 m is thin enough that it is influenced by the underlying material type. Therefore, with only 0.5 m of cover, having a coarser waste rock with less water retention capability will result in less evaporation and thus greater percolation.

3.9 Effect of Freshet Timing

Delaying the freshet by two weeks (Case #2) has a significant effect on the model outcome. This yields an average percolation of 14% with the range from less than 1 to 35%. The reason for this dramatic reduction is the fact that the model allows for evaporation starting on May 1, and therefore some of the freshet water can evaporate, rather than being allowed to simply percolate through the cover.

3.10 Effect of Sublimation

The model outcome is very sensitive towards the sublimation assumptions. Not allowing for any sublimation (Case #15) results in an increase in average percolation of 10% to 33%, while increasing the sublimation from 45 mm per year to 80 mm per year (Case #16) reduces the average net percolation from 23% to 14%. This result makes sense, as by increasing or decreasing the sublimation, the amount of water that is released during the freshet is increased or decreased, with freshet being a major contributor towards percolation in the Minto environment.

3.11 Effect of Wetter or Dryer Than Normal Climatic Years

To study the effect of abnormally dry or wet precipitation, synthetic subsets of five years were assembled, as described in Section 2.2.1. The models were run for 20 years, using the same 1,825-day weather record four times back-to-back. As expected, the dryer climate (Case #8) resulted in a significant reduction of the average percolation, to about 14%, with the range also shifting downward, to range between 8 and 26%.

The average percolation for the wetter climate (Case #7) remained unchanged compared to the base case, with the increase of the minimum percolation to 14% as expected. The decrease of the maximum percolation to 29% is however counter-intuitive, and is explained by the fact that although total yearly precipitation is higher in each of these years, there are fewer distinct high precipitation events driving massive percolation breakthrough.

3.12 Effect of Vegetation

The model indicates that grassy vegetation (pasture) established on the cover (Case #17) has a small effect on improving the cover performance. This result is perhaps less pronounced than general experience with vegetated covers would suggest. Evapotranspiration in cold climates is a complex process, and in the absence of site specific calibration data the modelled result remains indicative that any vegetation would modestly increase the cover performance.

4 Conclusions

The modelling results in this memo demonstrate that simple soils covers, using locally available materials can be effectively used to reduce percolation through the waste products (waste rock and tailings) by up to 50% compared to the uncovered waste. Uncovered, the waste rock and tailings percolation is about 39 and 45% respectively, while with a simple 0.5 m thick soil cover these average percolation rates decrease to about 23%. With some refinement, such as increasing the cover thickness to about 1 m and by adding some vegetation, the average percolation could likely be reduced to around 20% or less.

The sensitivity analysis shows that the outcome is most sensitive to climatic inputs (such as the timing of freshet and the rate of sublimation) as opposed to material properties, whether waste rock or cover soils. Since these climatic variables cannot be managed through engineering solutions, relying on the cover performance improvements suggested by these model outcomes would not be prudent.

It is important to note that in all cases, the overall range of percolation needs to be considered, as opposed to a single average value. Any water quality assessment should take into consideration the range of cover performance indicated by the modelling results as summarized in Figure 7.

Although the modelling presented has been verified by using a rigorous true surface flux boundary code, it remains uncalibrated and therefore the built-in conservatism is appropriate. One important example of this is evident in how the model is simulating runoff. The model results indicate that virtually no surface run-off will occur; however, there is clear evidence that in reality, runoff does occur at the site. Should covers be deemed an appropriate mitigation strategy, more rigorous refinement of the modelling should be considered at the more advanced design stages, including assessing 2-D effects of slopes to better reflect runoff and incorporation of the influence of frozen covers on freshet percolation. Notwithstanding the limitations mentioned in this document, the results as presented are deemed indicative of the most reasonable upper bound of expected percolation.

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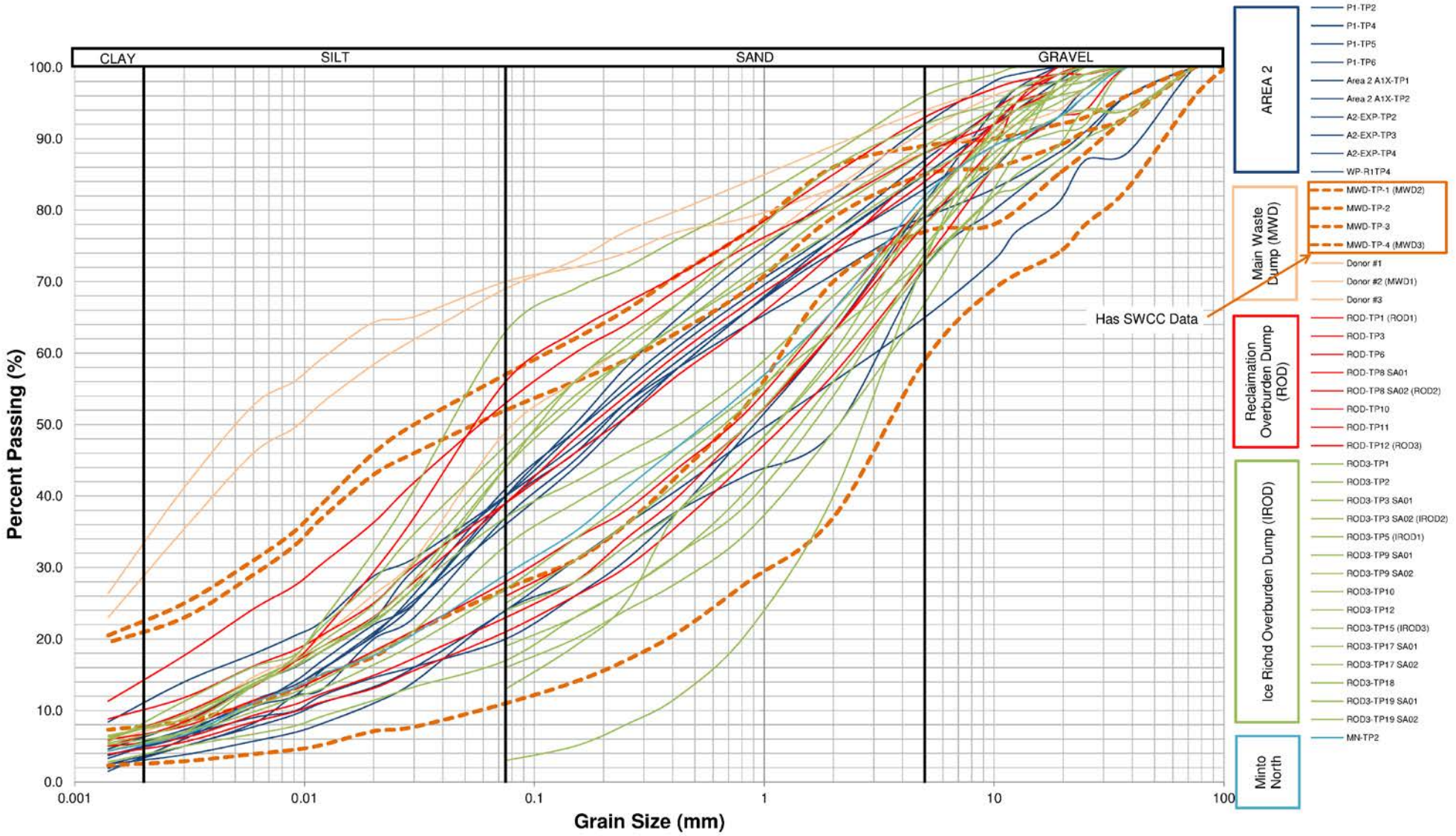
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Figures



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Closure Covers Model

Particle Size Distribution of all Candidate Cover Material on Site

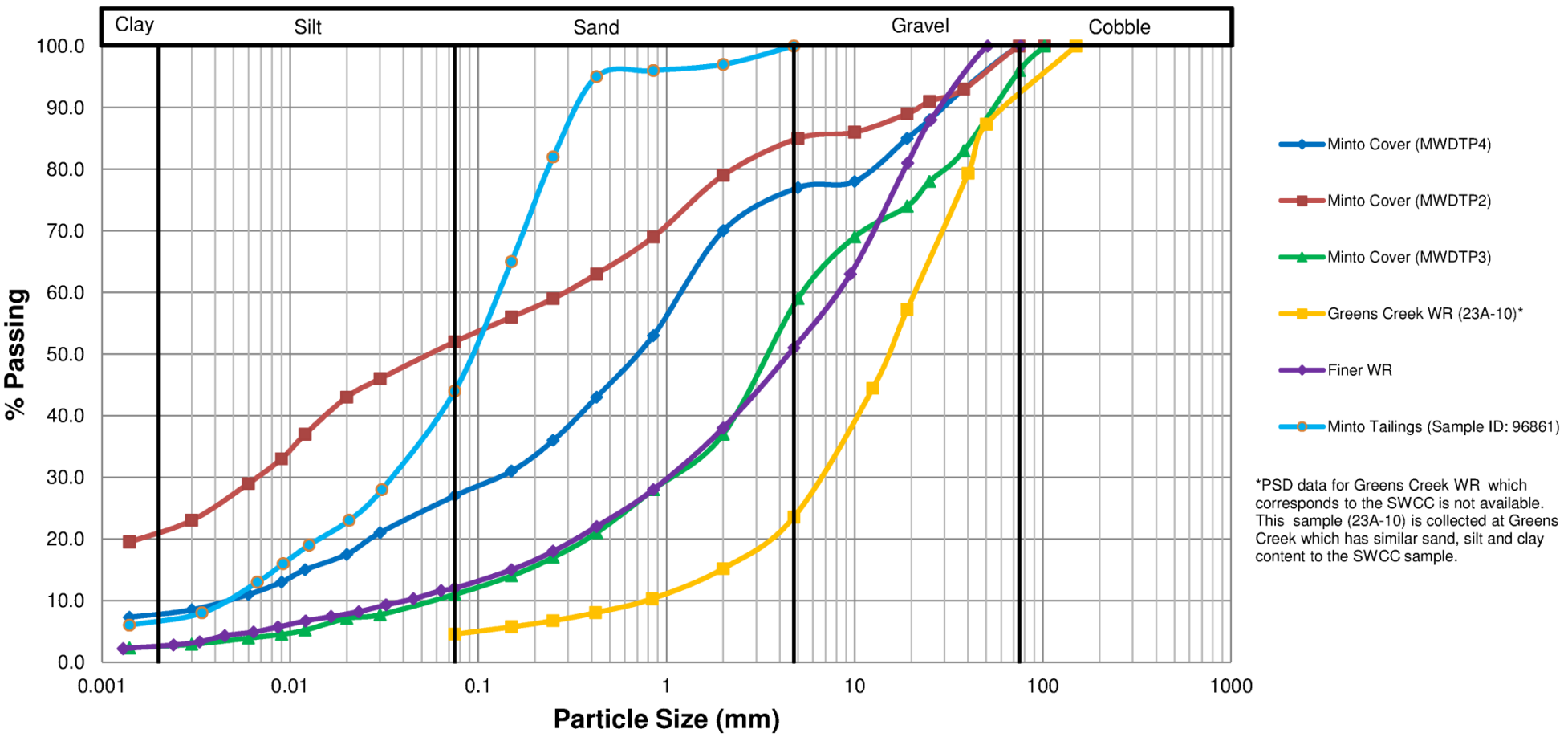
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*PSD data for Greens Creek WR which corresponds to the SWCC is not available. This sample (23A-10) is collected at Greens Creek which has similar sand, silt and clay content to the SWCC sample.

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Closure Covers Model

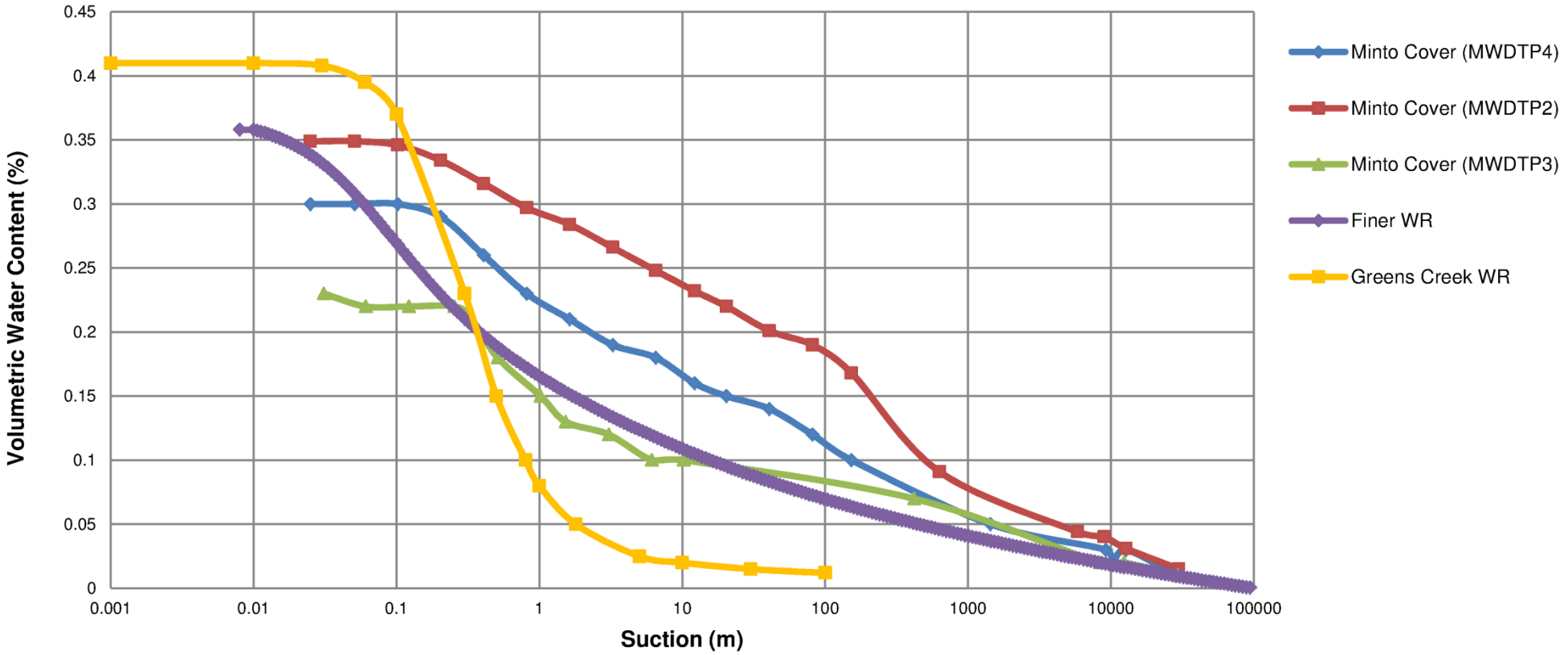
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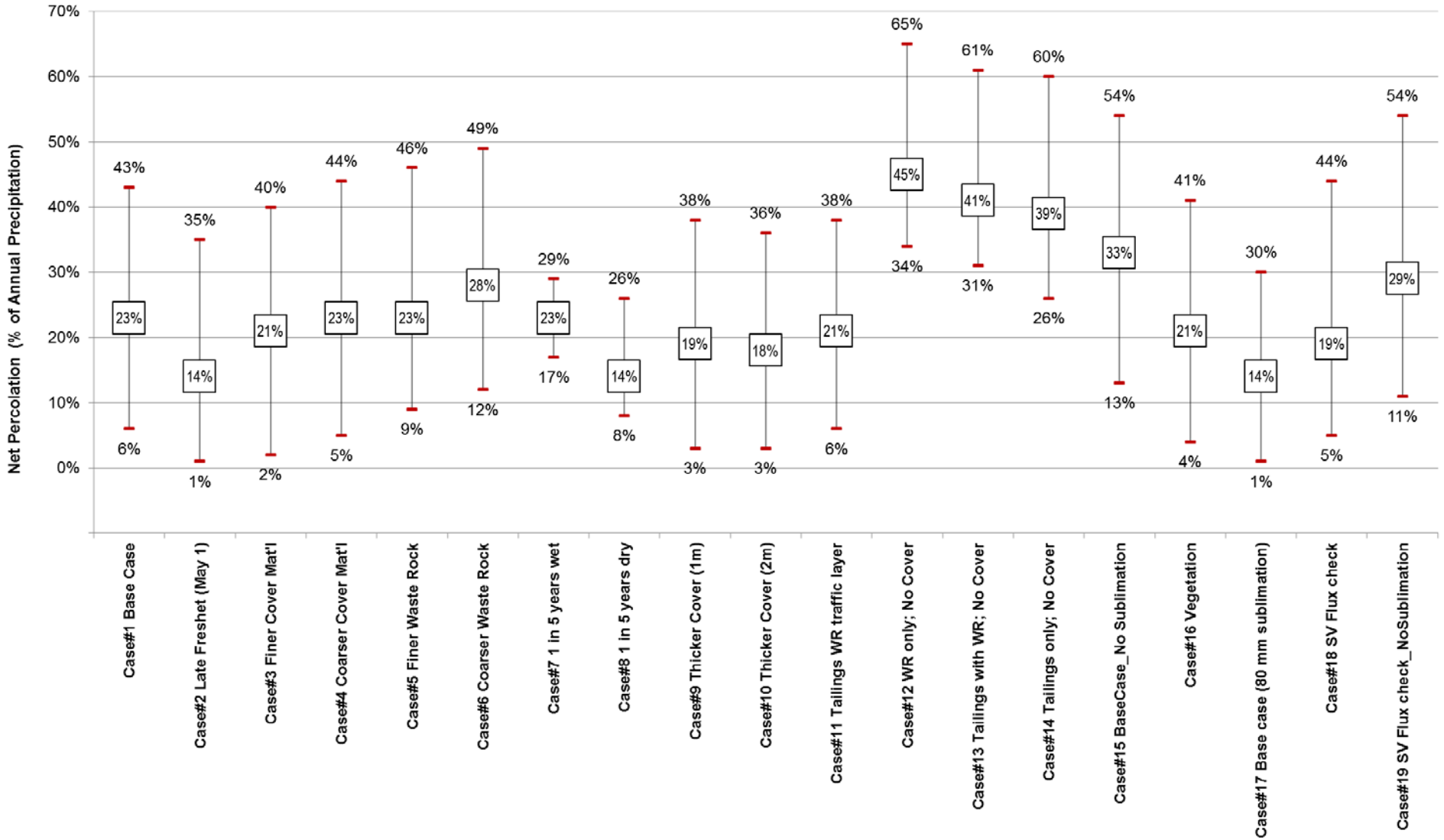
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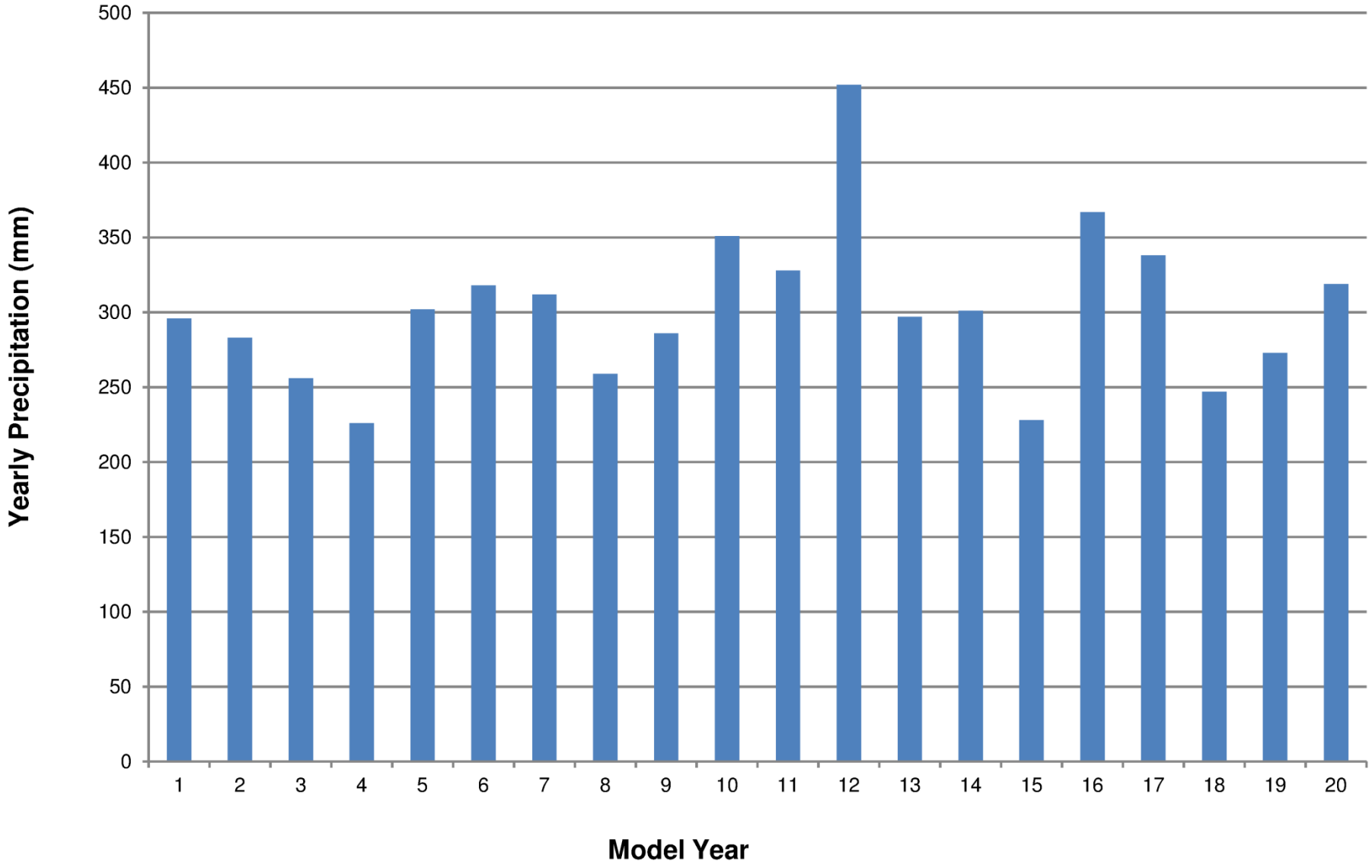
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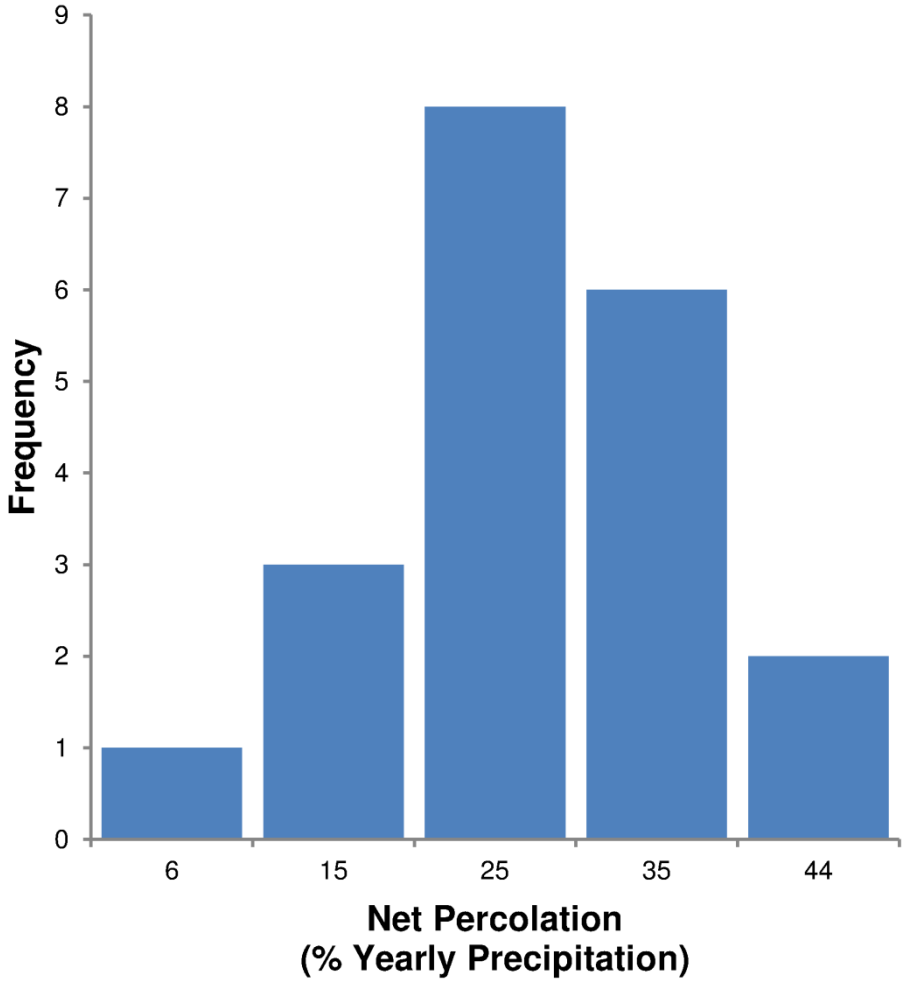
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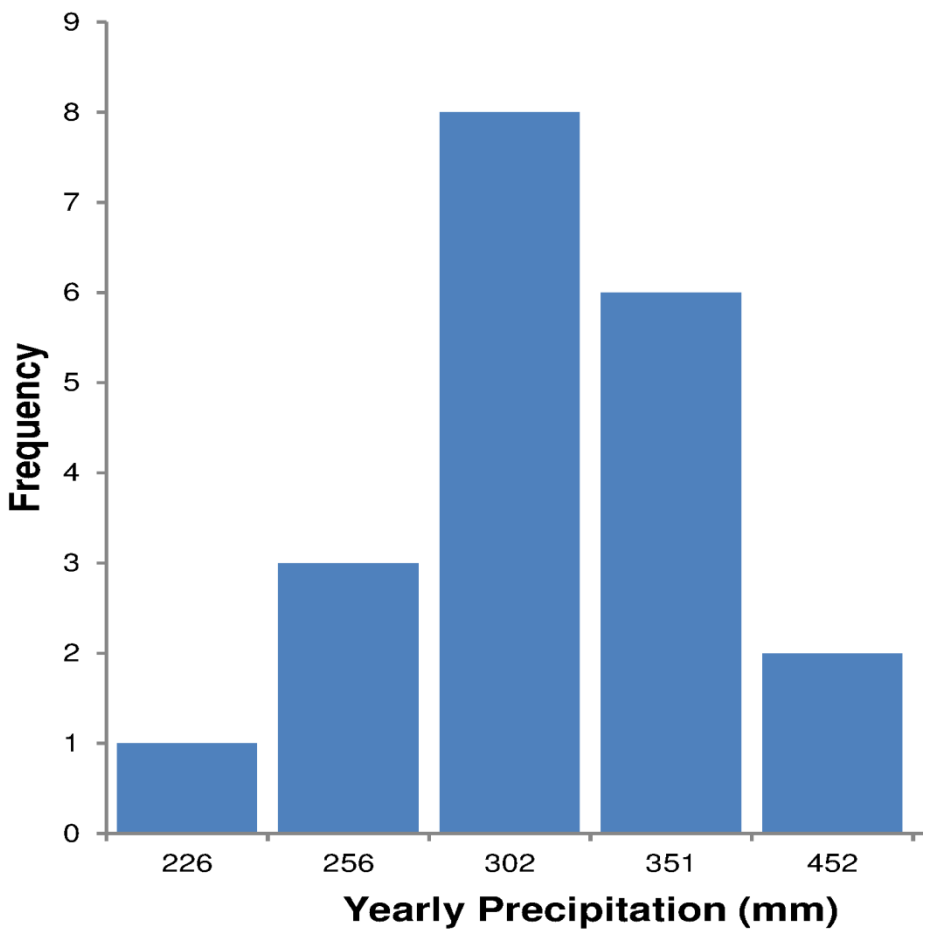
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		Date: Nov. 2014	Approved: IM
		Figure: 5	

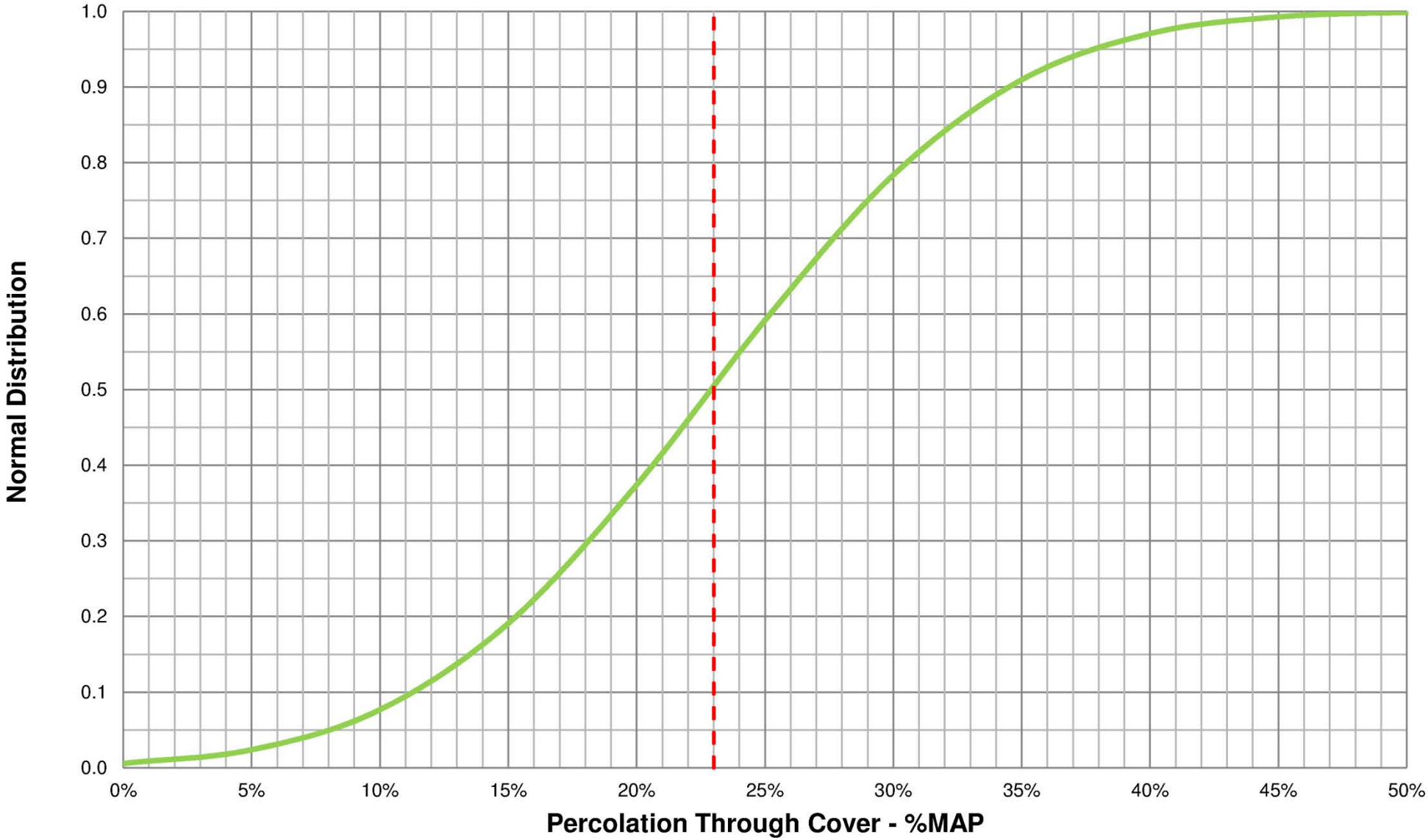
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Updated Closure Cover Design for the Minto Mine 2018 Reclamation and Closure Plan

Prepared for

Minto Explorations Ltd.



Prepared by

SRK Consulting (Canada) Inc.
1CM002.049
January 2018

Updated Closure Cover Design for the Minto Mine 2018 Closure and Reclamation Plan Update

January 2018

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Executive Summary

This report provides updated closure cover designs for the Southwest Waste Dump; Ridgetop Waste Dump; Ridgetop South Pit Backfill Dump; Ridgetop North Pit; Dry Stack Tailings Storage Facility; the Mill Valley Fill, and Mill Valley Fill Extension Stage 1 and 2; the Main Waste Dump, Main Waste Dump Expansion, and Main Waste Dump Wrap; Main Pit Dump; subaerial Main Pit tailings; and the Area 118 Backfill Dump. This document is considered the most current closure cover design, and has been completed in consideration of all previous work.

Several additional structures were incorporated into this version of the closure cover design, and aspects of the closure design have been advanced from previous work, most notably the hauling of cover materials from the Area 2 Stage 3 Pit, and the purchase of revegetation seed mixes.

This document shows the reader how a minimum cover thickness of 0.5 m was determined and, based on the re-grading plans presented, how it results in a total cover volume requirement of approximately 1,077,300 m^3; of which the majority of material has already been hauled to the Southwest Waste Dump, the DSTSF, the Main Waste Dump, and the Mill Valley Fill. Area 118 is also received off-spec overburden materials as backfill. The remaining cover volume is well within the estimated 2.3 Million bank cubic metres of cover material available for use from the reclamation overburden dump. The closure covers are to be constructed of material with no less than 10% silt and clay sized particles. In general, the re-grading of the facility was completed to provide a cut/fill balance targeting overall slopes as shallow as possible.

The stability analysis indicates that slopes shallower than 3H:1V are anticipated to be stable in the long term, but should slopes of 2.5H:1V be proposed, additional work is necessary to confirm these slopes will be stable.

The erosion analysis has concluded that bare (unvegetated) site cover material is highly susceptible to erosion on slopes; therefore, establishing a vegetated cover is important to the success of the cover. The proposed revegetation plan for slopes is anticipated to consist of seed mixes of native grasses, and application of fertilizer in support of establishing a strong vegetative cover to reduce the potential for sheet erosion and gully development on the cover. In areas where slopes are flatter, such as facility tops and benches, the revegetation plan is intended to include seeding of native plant communities.

The ability for each facility to shed and direct water was considered against a 1:200 year flood event. The approach adopted for the tops of structures was to limit the flow to less than 1 m^3/sec, such that armoured channels or swales would not be required, but that erosional protection could be achieved through vegetation. The flow directed over slopes was intended to be rip-rapped, with detailed specifications or riprap thickness to be established at a later date.

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1 Introduction

The Minto Mine is a high-grade copper mine located in the Yukon, approximately 240 km north of Whitehorse. The mine site occupies the valley in the upper reaches of Minto Creek, a tributary on the west side of the Yukon River. Operations began in October 2007 and are currently ongoing (2018).

Initially tailings were deposited in the Dry Stack Tailings Storage Facility (DSTSF) which was completed in 2012. Currently tailings and process water are managed in the Main Pit Tailings Management Facility (MPTMF) and the Area 2 Pit Tailings Management Facility (A2PTMF). In future, slurry tailings will be stored in the Ridgetop North Pit TMF, which will be utilized once mining is complete. Various waste rock storage facilities exist across the site including the Main Waste Dump (MWD), Main Waste Dump Expansion (MWDE), Main Pit Dump (MPD- active), Southwest Waste Dump (SWD), Ridgetop Waste Dump (RWD- planned), Area 118 Dump (active), and the Mill Valley Fill (and 2 extensions) (together, MVFE). Figure 1 illustrates these site facilities at the site.

1.1 Background

Numerous previous documents have been completed in support of various levels of closure planning. The most relevant document considered in support of this document include:

- Updated Closure Cover Design for the Minto Mine 2016 Closure and Reclamation Plan Update. SRK 2016c.
- Dry-stack Tailings Storage Facility Interim Cover Investigation. SRK 2016a.
- Closure Landform Design and Reclamation Landform Units for the Minto Mine. SRK 2016b.
- Minto Mine Closure Covers: Results of Numerical Modelling to Bracket Percolation Predictions. SRK 2015.
- Scoping Level Cover Assessment for Minto Closure Covers. SRK 2013a.
- 2012 Overburden Characterization Data Report for Minto Closure Covers. SRK 2013b.

This document has been prepared with the understanding that the above-mentioned reports are available to the reader and that should the reader choose, all relevant background to the cover designs is described.

Key conclusions from previous work have included:

- Cover materials placed without appropriate erosion protection are erodible and susceptible to gully development on the cover.
- The fine grained and mixed materials used to construct the interim cover of the DSTSF are suitable for use in the construction of final cover, but some revegetation efforts are necessary

to achieve cover functionality. The Residuum material is not a suitable stand-alone cover material as it cannot readily establish vegetation (SRK 2016a).

- The natural materials on-site are not expected to be capable of meeting the performance specifications of a low-permeability cover. Should a low-permeability cover be required, other options (including a geosynthetic) should be considered (SRK 2013a).

1.2 Advancements since the 2016 RCP

Minto has been advancing the progressive reclamation of the waste dumps with the direct hauling of overburden from the Area 2 Stage 3 Pit. Overburden materials that met the cover material specification (SRK 2016c) were hauled to:

- the Southwest Waste Dump
- the Main Waste Dump Expansion
- the Dry Stack Tailings Storage Facility, and
- the Mill Valley Fill Extension.

Materials that did not meet the design specification were wasted.

In addition to the direct hauling of overburden to the waste dump facilities, revegetation work continues to advance. Minto has purchased and is storing on-site sufficient seed to revegetate the slopes after the final cover placement has been concluded. The seed mix was specified on the basis that an initial vegetative cover could be developed to minimize the potential for erosion of the cover materials, and natural vegetation endemic to the region would establish over time.

The mine life has been extended to include the Area 2 Stage 3 pit, the Ridgetop North and South pits, and additional underground resources. This has resulted in modified designs (from SRK 2016c) to the Main Pit Dump, Main Waste Dump, the Area 118 Pit Backfill Dump, and the inclusion of the Ridgetop Waste Dump and the Ridgetop South Pit Backfill Dump. Minto also plans to mill the remaining high-grade waste portion of the Southwest Waste Dump[1], and to relocate SAT material currently stored in the Main Pit above elevation 786 m and place it below elevation 786 m so that it will be saturated over the long term.

1.3 Scope of Work

This report has been prepared to propose updated closure cover designs for the Southwest Waste Dump; Ridgetop Waste Dump; Ridgetop South Pit Backfill Dump; Ridgetop North Pit Tailings Management Facility; Dry Stack Tailings Storage Facility; Mill Valley Fill, Mill Valley Fill Extension Stage 1 and 2; Main Waste Dump, Main Waste Dump Expansion; Main Pit Dump and Main Pit Tailings Management Facility; and the Area 118 Backfill Dump. This document is considered the most current closure cover design, and has been completed in consideration of all previous work, while continuing to advance the level of detail in the engineering. This report

[1] A portion of the high grade waste originally stored in the Southwest Waste Dump has already been milled.

supersedes previous cover designs for the project including the designs proposed in previous closure plans.

This report is structured to lead the reader through descriptions of the closure cover objectives and criteria: the cover material characteristics, including overburden characterization, cover material thickness, erosion loss estimates, physical stability analysis, grading designs, water conveyance considerations, and re-vegetation descriptions for each of the above referenced facilities.

2 Objectives and Criteria

2.1 Design Objectives and Functions

It is currently proposed that the primary cover functions will be to:

- Reduce infiltration to the extent practical using locally available material;
- Ensure a stable landform that will promote establishment of natural vegetation endemic to the area; and
- Minimize ponding and surface erosion on the final landform.

2.2 Design Criteria

The proposed design criteria for closure covers at the Minto Mine are outlined in Table 1.

Table 1: Design Criteria

Component	Criteria	Comment/Rationale
Design Life	100 years.	Discussions related to closure often infer that closure measures (the soil cover, in this case) must last into perpetuity. While it is recognized that the cover will remain in place for a very long period, performance cannot credibly be measured in geologic timelines. Setting a realistic standard allows measurable targets to be set.
Oxygen Reduction	Not a defined criterion.	Oxygen reduction using covers is difficult to achieve in practice at the scale of full facilities. Instead, operational waste management practices have resulted in the majority of waste rock with NP:AP values below 3 being stored under conditions that will be saturated in the long term.
Infiltration Reduction	Not a defined criterion.	Minto and other project stakeholders wish to reduce infiltration (and net percolation) to the extent possible based on the available cover material's physical properties. The results of the cover modelling (SRK 2015) indicate that the net percolation is approximately 23%, but depending on the climactic conditions, can vary from 6% to 43%. Additional discussion is provided in Section 3.2.
Load Reduction	No specific target.	The site wide water and load balance assumes 20% net percolation through the cover over the long term, but acknowledges that net percolation (and therefore loading) may be greater or lower, depending on the climactic conditions.
Settlement	No criteria.	Settlement of the covers is not anticipated to occur as there is no settlement anticipated in the immediate foundation materials (waste rock, and compacted tailings). Long term thaw of permafrost foundations is expected to occur very slowly and to have little effect on cover performance.

Component	Criteria	Comment/Rationale
Seismicity	1 in 475-year recurrence interval.	Consistent with the BC Mined Rock and Overburden Piles Investigation and Design Manual, May 1991.
Physical Exposure	As far as practicable keep tailings and peripheral areas covered. At no time may more than 2.0 m^2 of mine waste be exposed as a contiguous area for the life of the structure.	It is recognized that the cover will evolve over time and factors such as extreme surface runoff beyond the stated design criteria, burrowing animals or human activities may cause damage to the cover.
Slope Stability	Overall factor of safety for the cover of 1.1.	Consistent with the BC Mined Rock and Overburden Piles Investigation and Design Manual, May 1991.
Wind Erosion	No criteria.	Based on site observations, wind erosion is not problematic.
Overland Surface Runoff	Capable of withstanding 1:200 year, 24-hour duration storm during peak freshet with no damage. An average soil loss of <6 tonnes/hectare /year is target from slopes, following the establishment and implementation of erosion mitigation strategies, such as vegetation. In no areas may overall cover thickness be less than 75% of original design thickness for the life of the structure.	Intend to construct a cover that could meet a soil erosion classification of Very Low (Wall et al. 2002)
Evapo-Concentration	No defined criteria.	Evapoconcentration is not expected to be an important post-closure process at Minto.
Root-Uptake	No defined criteria.	Root-uptake of porewater is not expected to be problematic, as there is no evidence of issues related to this on current revegetated areas.
Vegetation	Self-sustaining vegetation cover native to the region within 30 years of initial revegetation.	In areas of potential erosion concern (i.e. sloped cover facets), revegetation will focus on the rapid establishment of herbaceous ground-cover species using a mix of native grasses and one or more agronomic legumes, coupled with fertilizer applications in the early years (1-3) of establishment. In areas not targeted for the erosion-control treatment (i.e., landform plateaus and benches), revegetation will focus on re-establishing locally common native plant communities. Use of fertilizers on these areas will likely be avoided or minimized.
Land Use	General wilderness area. Large and small terrestrial animals, birds and aquatic life will be present. Humans will travel through the area infrequently (mostly hunters and trappers). Specific measures to preclude damage to the covers due to human and/or animal use is not required.	Wildlife habitat suitability will vary by vegetation type and structural stage as the vegetation becomes established and evolves.

Component	Criteria	Comment/Rationale
Landform	Promote use of landforms consistent with the current landscape. Provide for variability on cover thicknesses and landscaping as necessary to promote establishment of microclimates and variability.	The Closure Landform Design and Reclamation Landform Units for the Minto Mine will be used to guide the practical application of the landform design.

3 Cover Material Characteristics

3.1 Overburden Characterization

3.1.1 Geotechnical Characteristics

SRK evaluated 167 particle size distribution analyses and 38 Atterberg Limit (liquid and plastic limit) analysis completed on overburden samples at the Minto Site. Data was obtained from investigations completed, and documented by SRK (SRK 2013, 2016a&b). The data was reviewed, and indicated that in general the material can be variable from a particle size distribution perspective, with the amount of fines (less than 0.075 mm) in samples ranging between less than 5% to greater than 95%. Generally, the majority of the material that exists within the existing overburden stockpiles is expected to be in the range of 10% to 75% fines. Fines are used as a key indicator of the material as it can be used in the correlation of many soil parameters, including the moisture retention capacity of a soil. Typically, residuum samples were identified to have less than 10% fines. Residuum is a weathered bedrock material that has the consistency of a sand, and is typically a material that can be identified visually.

The Atterberg Limits are utilized to characterize the material in accordance with the Unified Soil Classification System (USCS) and the Modified Unified Soil Classification System (MUSCS). The material classifications generally range between CL-ML (Clayey Silt) to CL (Clay), which indicates the soils to have low to intermediate plasticity, and in the case of an ML (Silt), the behaviour is dominated by the silt sized particles present. The residuum material was deemed to be non-plastic.

For additional details related to the characterization of the geotechnical characterization of the overburden materials, refer to Appendix A.

3.1.2 Erosion

Soil erosion classification is based on the USDA soil textural classification. Table 2 provides a summary of the particle size diameter range based on the USCS / MUSCS, and the United States Department of Agriculture (USDA) soil textural classification.

Table 2: USCS vs USDA Particle Size Distribution Systems

Soil Component	Particle Size Diameter Rage (millimeters)	
	USCS / MUSCS	USDA
Boulders	> 200	n/a
Cobbles	200 – 76	n/a
Gravel	76 – 4.75	n/a
Sand	4.75 – 0.075	2.0 – 0.05
Silt	0.075 – 0.002	0.05 – 0.002
Clay	< 0.002	< 0.002

The samples were re-classified, and plotted on a soil texture triangle to determine the general soil texture. The samples typically categorized as sandy loam. For additional details related to the characterization of the soil classification of the USDA methodology, please refer to Appendix A.

3.1.3 Quantities

The reclamation overburden dump (ROD) contains approximately 2.3 Million bank cubic metres (EBA, 2010) of overburden. The overburden was characterized prior to mining, and a summary of the material properties is provided in Appendix A. The PSD of the material recovered from Area 2 is listed in Tables 2 and 3. Generally, the overburden material was not selectively placed within the ROD during placement, and material that would be excavated from the ROD for cover purposes is expected to be somewhat variable in quality, but with greater than 10% fines.

The overburden characterization from the Area 2 Stage 3 Pit indicated that the material is similar to the overburden currently contained in the ROD, but with some samples containing higher degrees of clay and silt. Based on the investigation completed, there were no mineable units within this area that were discretely targeted to produce material of higher or lower fines content. A more detailed description of the investigation and results is provided in Appendix B.

In 2014 overburden fine grained soils and residuum was place on portions of the DSTSF as a trial overburden cover. Those areas capped predominately with residuum did not support vegetation, and therefore did not meet the cover objectives. Areas of fine grained soils, and areas of mixed fine-grained soils and residuum do support vegetation and therefore do meet the cover objectives.

3.2 Cover Thickness to Achieve Net Percolation Targets

SRK completed numerical modelling to bracket percolation predictions in 2015 (SRK 2015). The modelling indicated that without a cover, the net percolation could be between 39% percent and 45%, and depending on climactic conditions, could vary between 26% and 65%. Cover materials were applied, with the base case cover material consisting of material properties from sample MWD-TP4 collected at the Minto Mine, which is a gravel and sand material with more than 25% fines (<0.075 mm). The results indicate that the net percolation is approximately 23% (under Mean Annual Precipitation (MAP) conditions), but depending on the climactic conditions, can vary from 6% to 43%. The cover material was varied and an analysis was also completed considering a coarse material (MWD-TP3), which is predominantly gravel and sand with less than 10% fines (<0.075 mm). The results of this analysis indicate that the net percolation was approximately 23%, but depending on the climactic conditions, can vary from 5% to 44%.

The modelling results indicate that there is roughly a 50% decrease in net percolation (from around 40% MAP to around 20% MAP) after cover materials are included in the analysis. The cover thickness was assumed to be approximately 0.5 m thick, and following sensitivity analysis to the thickness (1 m and 2 m) only minor additional decreases in net percolation were projected. Therefore, a minimum cover thickness of 0.5 m was adopted for the project, with a requirement that the materials contain greater than 10% fines.

3.3 Erosion Loss Estimate

The purpose of this analysis was to present the potential effects of erosion due to sheet and rill water erosion that could occur on the engineered slopes at Minto, and evaluate a range of conditions and parameters to help guide the landform designs. Sheet and rill erosion occurs as a result of flows that are not concentrated into a particular flow path, but over time, if allowed to persist, can develop into larger erosion features such as gullies. The intent is then to determine which methods of erosion protection are sufficient to reduce erosion to acceptable levels, to minimize the potential for development of gullies, and to characterize what (if any) sacrificial thickness should be added to the cover to account for erosion. This section provides a summary of the erosion loss analysis completed, with additional details found in Appendix C.

Erosion that may occur within channel flow and the associated armouring is not considered in this discussion.

Based on the RUSLEFAC equation, SRK has targeted a soil erosion classification of “Very Low” which means that an acceptable rate of erosion loss is approximately 6 Tonnes per hectare per year (Wall et al. 2002). Soils with erosion classifications of “Very Low” demonstrate slight to no erosion potential. Minimal erosion problems should occur if good soil conservation management methods are used. A tolerable soil loss (<6 T/ha/year) is the maximum annual amount of soil which can be removed before the long term natural soil productivity of a hillslope is adversely affected (Wall et al. 2002). By targeting a soil erosion classification of “Very Low”, the intent is to limit the development of rill, inter-rill, and ultimately gully erosion. In some cases, a soil can meet this target on its own, but in many cases, as is the case at the Minto site, support practices are required to achieve this target.

As described in Appendix C, soil loss over the course of the design life (100 years) was calculated to determine whether the average depth of soil loss would reduce the initial cover thickness to below the required cover thickness. The soil loss was calculated for unvegetated (bare) soils, and for soils with 80% coverage with short rooted plants. As is observed on the vegetation trials on the Main Waste Dump, under appropriate conditions, this can easily be achieved.

Annual soil loss due to water erosion was multiplied by 100 years to determine design life soil loss, which is presented for several straight slope scenarios in Table 3. Construction of complex slopes was calculated to have a potential impact on decreasing erosion by approximately 10%, which is discussed further in Appendix C. Average annual soil loss (in T/ha/year) is also presented in Table 3.

Table 3: Calculated Water Erosion Design Life Soil Loss

Slope Condition		Design Life Soil Loss (mm) per Slope Length							
		50 m		85 m		100 m		150 m	
		Annual (T/ha/yr)	100 yrs (cm)	Annual (T/ha/yr)	100 yrs (cm)	Annual (T/ha/yr)	100 yrs (cm)	Annual (T/ha/yr)	100 yrs (cm)
Non-Vegetated	**2.5H:1V**	31.8	19.9	41.8	26.1	45.6	28.5	56.2	35.1
	3H:1V	26.2	16.4	34.1	21.3	37.1	23.2	45.5	28.4
	3.5H:1V	22.1	13.8	28.6	17.9	31.0	19.4	37.8	23.6
	4H:1V	19.0	11.9	24.3	15.2	26.4	16.5	31.9	19.9
	5H:1V	14.6	9.1	18.4	11.5	19.8	12.4	23.7	14.8
Vegetated (80% Short-Rooted Plant Coverage)	**2.5H:1V**	3.7	2.3	4.8	3.0	5.3	3.3	6.5	4.1
	3H:1V	3.0	1.9	3.9	2.5	4.3	2.7	5.2	3.3
	3.5H:1V	2.6	1.6	3.3	2.1	3.6	2.2	4.4	2.7
	4H:1V	2.2	1.4	2.8	1.8	3.0	1.89	3.7	2.3
	5H:1V	1.7	1.1	2.1	1.3	2.3	1.4	2.7	1.7

Table 3 illustrates the value of vegetation, and therefore the establishment of vegetation on the slopes is critical to the success of the closure covers. Short term support practices will be required to develop a good vegetated cover. Examples of short term support practices are rolled erosion control products, slope texturing, and hydro-seeing with an erosion resistant tackifier. The potential effect that short term support practices have on reducing cover erosion are discussed in further detail in Appendix C. Once vegetation is established, the calculated soil loss is less than 5 cm, which is generally within the placement tolerance of earthworks when using large equipment and is therefore considered both negligible and acceptable.

3.4 Physical Stability

SRK evaluated the cover stability of the closure covers at various slope angles to identify if some of the available borrow material is better suited to some areas versus others. The global stability of the operational design of each of the waste facilities, under both operations and closure conditions, have been previously evaluated in their respective design documents. Generally, closure configurations will result in re-sloped/landscaped configurations with shallower slopes. Global stability is not considered further in this analysis, but may be considered further following the development of final re-grading plans discussed in this document.

Generally, the physical stability of a cover is a function of the normal stress over the cover, the internal shear strength of the cover material, the interface shear strength between the underlying material and the cover material, as well as the seepage forces present within the cover. In the case of most of the Minto cover designs, a lower strength cover material is proposed to be placed over a material with higher strength – either compacted tailings, the compacted shell of the DSTSF, or waste rock. Therefore, the critical failure mode is a failure that occurs along the interface of the two materials, and is controlled by the shear strength in the weaker cover material.

SRK completed a 2-dimensional limit equilibrium analysis stability analysis as described in Appendix D. The analyses were focused on base case scenarios where a piezometric surface was placed midway through the cover thickness, and the material underlying the cover was considered impenetrable to force the cover failure either through the cover material, or along the interface. Two sets of analyses were completed to demonstrate the effect of a cover constructed of residuum material versus a cover constructed of silty sand material. Analyses were completed for slopes of 2.5H:1V, 3H:1V, 4H:1V, and 5H:1V. Various sensitivity analysis was completed, which indicated that the models were most sensitive to the piezometric surface in the cover.

All of the base case scenarios met the minimum target factor of safety with the exception of the placement of a silty sand cover material on a 2.5H:1V slope. Additional analyses to evaluate the impact of a variable piezometric surface in the cover placed over a 2.5H:1V slope were completed as the base case conditions for the silty sand cover did not meet the minimum target factor of safety. The results indicated that increases in the piezometric surface decreased the factor of safety below the base case, while decreases in the piezometric surface increased the factor of safety above the target criteria.

The analysis concluded:

- Residuum material is preferable to be placed on a 2.5H:1V slope from a geotechnical stability perspective. Erosional susceptibility of the material (as discussed in Appendix C) and its ability to support re-vegetation efforts should be considered prior to final selection of cover material. It should also be noted that currently the Main Waste Rock Dump has 2.5H:1V slopes, and has been covered with silty sand material, and vegetation trials are on-going. The cover material does not appear to be prone to continued sloughing or cover failure, and it is likely that the waste rock below drains the cover and limits the potential for the piezometric level to increase to such a point that the seepage forces influence the cover stability below unity. Vegetation on the cover varies from well covered to sparsely covered.
- Slopes of 3H:1V, or shallower are not restricted to the type of cover material based on geotechnical performance.

3.5 Design Parameters

Table 4 provides a summary of proposed design parameters to be adopted for the Minto Closure Covers, based on the particle size distributions, and the net percolation cover modelling completed (SRK 2015).

Table 4: Closure Cover Design Parameters

Description		Value
Cover Thickness		0.5 m (minimum)
Cover Material Specifications	Gravel	0 % to 40%
	Sand	60% to 90%
	Fines	> 10%
Soil Texture Classification	Sandy Loam	Sandy Loam

4 Revegetation

The preliminary revegetation plan is designed to achieve the land-use objectives of wildlife habitat re-creation, creation of habitat for traditionally used plants, and the return of biodiversity values over time. The primary focal wildlife species is moose, although re-creation of habitat for prey species such as snowshoe hare and upland game birds may be possible. Revegetation treatments are designed to achieve these objectives, and also to perform the key task of protecting the placed cover materials from erosion where required. Classification of the revegetation objectives are as follows:

1. Slopes versus plateaus – a primary goal of revegetation is to reduce erosion, where needed. Thus, we identified sloped areas for erosion-control revegetation treatments and level areas where these treatments are not required; and
2. Zonal(mesic) sites versus drier (subxeric-submesic) sites – the former occur on tailings substrates and north-facing slopes, while the latter occur on waste-rock substrates and south-facing slopes.

Mapping of the area indicated that approximately 42 ha of level/plateau areas and 38 ha of sloped areas will be ready for revegetation in 2018. Seed mixes were designed for these areas based on the following principles, with the overarching objective that revegetation materials should be wholly or mostly made up of species native to the Yukon:

- **Sloped areas** – the primary revegetation goal on these areas is erosion control, so the seed mix should be primarily composed of native grass species, seeded relatively heavily (35 kg/ha), with an additional component of annual ryegrass and alfalfa, to accelerate revegetation and reduce the amount of bare ground exposed to erosion. Alfalfa was included as it is the only native legume still listed as an acceptable species by the Yukon Revegetation Manual (Matheus and Omtzigt 2013).
- **Level areas** – the primary revegetation goal on these areas is promotion of recovery towards pre-development ecosystems. For this reason, the primary revegetation strategy is natural regeneration, as this has been observed on many sites at Minto, and as the relatively small development footprint and proximity of non-mined seed sources facilitates this recovery mechanism. However, observations on the MWD revegetation trial and the naturally revegetating overburden stockpiles suggest that foxtail barley is present and readily able to colonize sites. This species is undesirable as a large component of revegetation, as it can cause problems for grazing wildlife after seed formation, as the long seed awns can get stuck in the mouth, nose, and eyes of grazing animals, potentially causing irritation and infection. Therefore, a seed mix was developed for level areas at Minto, with the goal of providing some occupation of space to prevent full colonization by foxtail barley, but also of leaving space for natural colonization of these sites. The level-ground seed mix is composed solely of grass species native to the Yukon, applied at a rate lighter (20 kg/ha) than that used for the sloped areas.

Yukon Energy, Mines and Resources (2013b) defines re-vegetation as the re-establishment of vegetation on land which previously had vegetation cover. The objective of revegetation of

mining disturbances is "*to leave the ground in such a way as to provide a good chance for successful re-vegetation by plant species native to the site and the area (natural revegetation)".* It is anticipated that the two primary revegetation treatments proposed above for Minto are consistent with this over-arching objective. Native species tend to be slower to establish than agronomics (which have been selectively bred for rapid establishment), and with the high silt content and erosion susceptibility of the Minto cover materials, rapid revegetation is critical to minimize erosion and protect the cover material. Restriction to use of native species only would likely result in high erosion rates, loss of cover materials through sheet and/or rill/gully erosion, and likely subsequent poor revegetation. Targeted use of agronomics will promote cover stability at the ground surface, and will be more successful in eventually establishing native species. As excessive erosion would be a key failure mode for the cover system, the revegetation treatments have been developed to give priority to reduction of risk of this failure mode. A transition to native species on these erosion-control areas is anticipated to occur naturally, due to the focus on native-species establishment on other mine areas and due to the proximity of surrounding vegetation-propagule sources in adjacent intact ecosystems.

Additional detail on development of revegetation treatments, and associated recommendations, are provided in the technical memo attached as Appendix E. Minto has purchased, and has on-site seed to apply to the covers after it is determined that the minimum cover thickness has been placed. This is planned to occur in 2018.

5 Re-grading Designs

5.1 Southwest Waste Dump

Progressive reclamation of the Southwest Waste Dump began in 2015. Minto re-sloped the face of the dumps to slopes ranging roughly between 12H:1V and 4H:1V and has hauled and stockpiled overburden necessary to cover and revegetate the top of the dump, and the sump slopes. Remaining work to be completed includes:

- Spreading and grading of the overburden at the top of the dump, and the slopes;
- Grading on the top surface to shed water into swales designed to carry the flow down the face of the dump without causing significant erosion;
- Excavation of the high-grade waste stockpile (for either milling or relocation);
- Additional design work;
- Construction of swales;
- Cover placement; and
- Detailed planning and implementation of revegetation plans.

A minimum cover thickness of 0.5 m is proposed for the Southwest Waste Dump across the facility, including the footprint of the high-grade waste stockpile. Localized areas may require additional grading to meet design grades, and it is proposed that final grades will be achieved through the placement of overburden, subject to Minto's scheduling plans.

To shed water, the top of the medium grade waste area is proposed to be split into six catchment areas while the bulk waste area is proposed to be split into three small catchment areas. These catchment areas are illustrated in the detailed hydraulic analysis information contained in Appendix F. The intent is to minimize the amount of water that flows directly over the slope as sheet flow. Each of these small catchments would be constructed to direct water into a broad swale, currently considered to be 2 m wide at the base, and have side slopes of 10H:1V. The swales were designed based on contributing watershed area, and flow depths are anticipated to range between 0.13 m and 0.37 m during a 1:200 year 24-hour flood event. The reported depths are double that of the calculated depths to allow for ice accumulation; however, this is a conservative element, as no significant ice accumulation is anticipated given the expected unsaturated nature of the underlying waste rock and the lack of winter groundwater discharge to the channels.

The swales have been designed to flow over and maintain the same base with the slope, but transition to 3H:1V side slopes along the slopes. The swales on the top are proposed to be protected with vegetation, while the slopes are proposed to be armoured with a gravel to cobble sized rip rap. Rip rap thickness and final dimensions are yet to be determined. Details regarding the hydraulic designs of these swales are provided in Appendix F.

The proposed re-grading and swale locations are illustrated on Figure 2. The Southwest Waste Dump is expected to require approximately 322,200 m^3 of cover material. Of this, sufficient overburden has been hauled to the top of the mid-grade pile and stockpiled to achieve final grades, and to apply along the slope. It is estimated that over 60% of the material necessary to cover the Southwest Waste Dump has been hauled and stockpiled. The proposed slopes are 3H:1V or shallower, and therefore physical stability of the cover is not anticipated to be problematic.

5.2 Ridgetop Waste Dump

The design of the Ridgetop Waste Dump final grading is based on the permitted design of the dump. As construction of the dump has not yet started, the configuration of the as-constructed dump is not yet known. Remaining work to be completed includes:

- Additional planning and design work, using as-built dump surface details;
- Cover placement; and
- Detailed planning and implementation of revegetation plans.

Consistent with the Southwest Waste Dump design, a minimum cover thickness of 0.5 m is proposed across the facility. Localized areas may require additional grading to meet design grades, and the decision to achieve final grades by grading waste rock, or using overburden will be determined based on Minto's scheduling plans.

To shed water, the top of the Ridgetop Waste Dump is proposed to be split into three small catchment areas that direct water away from the slopes of the dump towards natural ground at the eastern side of the dump. These catchment areas are illustrated in the detailed hydraulic analysis information contained in Appendix F. The intent is to minimize the amount of water that flows directly over the slope as sheet flow. Each of these small catchments would be constructed to direct water into a broad swale, currently considered to be 2 m wide at the base, and have side slopes of 10H:1V. The swales were designed based on contributing watershed area, and flow depths are anticipated to range between 0.18 m and 0.37 m during a 1:200 year 24-hour flood event. The reported depths are double that of the calculated depths to allow for ice accumulation; this is conservative, as explained in Section 5.1 for the Southwest Waste Dump. The swales on the top are proposed to be protected with vegetation. Details regarding the hydraulic designs of these swales are provided in Appendix F.

No swales have been designed along the slopes of the dumps as there is no designed catchment to route along the slopes of the dump.

The proposed re-graded facility is illustrated on Figure 2. The Ridgetop Waste Dump is estimated to require approximately 194,500 m^3 of cover material. The proposed slopes are 3H:1V or shallower, and therefore physical stability of the cover is not anticipated to be problematic.

5.3 Ridgetop South Pit Backfill Dump

The design of the Ridgetop South Pit Backfill Dump final grading is based on the permitted design of the dump. As excavation of the pit, and construction of the dump has not yet started, the configuration of the as-constructed dump is not yet known. Remaining work to be completed includes:

- Additional planning and design work, using as-built dump surface details;
- Cover placement; and
- Detailed planning and implementation of revegetation plans.

Consistent with the other waste dump designs, a minimum cover thickness of 0.5 m is proposed across this facility. Localized areas may require additional grading to meet design grades, and the decision to achieve final grades by grading off-spec overburden, or using suitable cover material will be determined based on Minto's scheduling plans. It may be determined that suitable cover material for this dump will be residuum, which is consistent with the natural soils surrounding the dump.

To shed water, the top of the Ridgetop South Pit Backfill Dump is anticipated to shed away from the dump, to the west. An estimate of the catchment areas and swale sizing is provided in the hydraulic analysis information contained in Appendix F. Consistent with the designs for the other waste structure, the catchment would be constructed to direct water into a broad swale, currently considered to be 2 m wide at the base, and have side slopes of 10H:1V. The swale was designed based on contributing watershed area, and the flow depth was estimated to be 0.22 m during a 1:200 year 24-hour flood event. The reported depth is double that of the calculated depths to allow for ice accumulation; this is conservative, as explained in Section 5.1 for the Southwest Waste Dump. The swales on the top are proposed to be protected with vegetation. Details regarding the hydraulic designs of these swales are provided in Appendix F.

The Ridgetop South Pit Backfill Dump design has not been finalized. However, to ensure that here is appropriate cover material allocated for scheduling purposes, the proposed re-grading design was used to estimate the volume of cover material that may be required. The Ridgetop South Pit Backfill Dump p is estimated to require approximately 16,500 m^3 of cover material. The proposed re-grading is illustrated on Figure 2.

5.4 Ridgetop North Pit Tailings Management Facility

The Ridgetop North Pit has not yet been mined, and therefore the final configuration is subject to change based on the as-built condition. It is proposed that the Ridgetop North Pit will be backfilled with tailings, and referred to as the Ridgetop North Pit Tailings Management Facility. Final closure configuration of the Ridgetop North Pit does not include a water cover, and therefore a soil cover will be required to be placed over the tailings. It is proposed that the cover will consist of a rock trafficking layer and a 0.5 m thick overburden cover; however, detailed engineering has not been advanced for this concept. The rock trafficking layer is also intended to reduce the potential for boils to develop in the cover.

A preliminary plan for re-grading is illustrated on Figure 2. The proposed plan assumes that the cover will be placed at (or near) elevation 862 m, and that it will grade towards a central swale designed to carry water off of the facility. The broad swale is currently considered to be 2 m wide at the base, and have side slopes of 10H:1V. The swale was designed based on the contributing area of the Ridgetop North Pit, and is estimated to have a total depth of flow of 0.48 m. The reported depth is double that of the calculated depth to allow for ice accumulation; this is conservative, but as the tailings are expected to impede drainage of the cover, there is a higher probability of ice accumulation within this swale.

The re-grading plan for the facility accounts for an allowance of 1 m of trafficking layer, as it is anticipated some of the trafficking layer will be lost into the tailings, and 0.5 m of cover material. This results in a need for approximately 57,000 m^3 of material to provide the trafficking surface, and 28,500 m^3 of cover material to place over the trafficking surface.

5.5 Dry Stack Tailings Storage Facility

Progressive reclamation of the DSTSF completed to date has consisted of placement of an interim cover over the tailings. This served multiple purposes, including isolating the tailings from wind and water erosion, as well as informing the success of using different cover materials. Observation of the interim cover has led to the conclusion that residuum material on its own is not a preferred cover material; however, a mix of residuum and fine-grained overburden will be able to achieve the overall closure cover objectives. Select areas of the cover have been excavated to achieve the desired drainage grades, without adding additional mass (in cover material) that may detrimentally impact the stability of the DSTSF. In addition, the downstream shell of the DSTSF has been re-graded on the eastern side, and approximately 2/3 of the downstream shell has been re-graded along the northern side. Overburden material was stockpiled at the crest of the dump, and has been spread along areas where the dump face has been re-graded. Remaining work to be completed includes:

- Re-grading of the western end of the rock shell face of the DSTSF;
- Re-grading and surface material amendment on the top of the DSTSF;
- Confirmation that the surface material amendment placed meets the minimum cover thickness requirements;
- Additional design work related to final local drainage network details; and
- Detailed planning and implementation of revegetation plans.

A minimum cover thickness of 0.5 m is proposed for the DSTSF. To shed water, the top of the DSTSF is split into five catchment areas, intending to direct water off the facility to the west, north, and south-east. These catchment areas are illustrated in the detailed hydraulic analysis information contained in Appendix F. The intent is to minimize the amount of water that flows directly over the slope as sheet flow. Each of these small catchments would be constructed to direct water into a broad swale, currently considered to be 2 m wide at the base, and have side slopes of 10H:1V. The swales were designed based on contributing watershed area, and flow

depths are anticipated to range between 0.12 m and 0.27 m during a 1:200 year flood event. The reported depths are double that of the calculated depths to allow for ice accumulation; this is conservative, as explained in Section 5.4 for the Ridgetop North Pit Tailings Management Facility.

Only one swale is proposed along the slopes of the DSTSF. This is proposed along the northern slope, and designed to carry water down the face of the DSTSF and onto the Mill Valley Fill Extension. Where this swale transitions from the top of the DSTSF to the face of the DSTSF, it will transition to a 2 m wide base channel with 3H:1V side slopes. The swales on the top are proposed to be protected by vegetation, while the swale on the slope is proposed to be armoured with a gravel to cobble sized rip rap. Rip rap thickness and final dimensions are yet to be determined. The proposed re-grading and swale locations are illustrated on Figure 2.

Soil loss due to erosion caused by overland sheet flow (as discussed in Section 3.3) has the ability to impact the integrity of the cover and it is proposed that the revegetation concepts described in Section 4 be implemented. The proposed slopes are 4H:1V or shallower, and therefore physical stability of the cover is not anticipated to be problematic.

The estimated cover volume necessary to cover the DSTSF is 111,900 m^3. All of this volume has been hauled to the DSTSF, and needs to be spread across the DSTSF top surface and shell, and confirmation of the final thickness to be completed thereafter.

5.6 Mill Valley Fill Extension Stage 1 & 2

The Mill Valley Fill Extension Stage 1 and 2 were designed with closure in mind. The top surfaces were generally graded to shed water towards the north.

There are three main terraces to the waste dump. The first westernmost terrace will require some effort to re-grade so that water is directed to the west, rather than to the north. The middle terrace is proposed to be graded north and will also convey water shed off the DSTSF (through constructed swales). The eastern terrace is proposed to be graded to shed water off to the north-west. The downstream slopes of the Mill Valley Fill Extension have been completed and closure cover placement has already begun. Closure cover placement and spreading has also begun on the top surfaces of the dumps. Remaining work to be completed includes:

- Cover placement along the slopes;
- Cover grading of the top dump surfaces;
- Swale construction;
- confirmation that the surface material placement placed meets the minimum cover thickness requirements;
- additional design work related to final local drainage network details; and
- detailed planning and implementation of revegetation plans.

Consistent with other facilities, a minimum cover thickness of 0.5 m is proposed and localized areas may require additional grading to meet design grades. Design grades may be achieved through the placement of either waste rock, or overburden based on Minto's scheduling plans.

To shed water, the top of this facility was separated into three small catchment areas, controlled by the surface area of each dump terrace. These catchment areas are illustrated in the detailed hydraulic analysis information contained in Appendix F. Consisted with the other facilities, the intent is to minimize the amount of water that flows directly over the slope as sheet flow. All three of these small catchments would be constructed to direct water into a broad swale, currently considered to be 2 m wide at the base, and have side slopes of 10H:1V. The swales were designed based on contributing watershed area, and flow depths of approximately 0.19 m to 0.39 m during a 1:200 year 24-hour flood event. The reported depths are double that of the calculated depths to allow for ice accumulation; this is conservative, as explained in Section 5.1 for the Southwest Waste Dump.

Consistent with the design philosophy employed with the other facilities, the swales have been designed to flow over the slope and transition to a channel base width equal to that of the swale on the top surface with 3H:1V side slopes. The swales on the top are proposed to be protected with vegetation, while the slopes are proposed to be armoured with a gravel to cobble sized rip rap. Rip rap thickness and final dimensions are yet to be determined. The proposed re-grading and swale locations are illustrated in Figure 2.

As discussed in Section 3.3, soil loss due to erosion caused by overland sheet flow can impact the integrity of the cover. However, it is proposed that the revegetation concepts described in Section 4 be implemented on the slopes to minimize erosion, and increase the rate of success for revegetation. The proposed slopes are 3H:1V or shallower, and therefore physical stability of the cover is not anticipated to be problematic.

The proposed plan includes approximately 71,300 m^3 of cover material, all of which has already been hauled to the dump.

5.7 Main Waste Dump and Main Waste Dump Expansion

Progressive reclamation began at the Main Waste Dump through the placement of cover material, and vegetation trials on two portions of the re-sloped benches. The trials have illustrated various level of success, but reinforces the conclusions from the erosion analysis: that without vegetative support, when placed on slopes the overburden material is highly susceptible to rill erosion, which has led to the development of gullies along the face where vegetation has not been successful. The re-graded slopes were re-graded to 2.5H:1V. An extension of the Main Waste Dump toe is currently under construction with waste rock, and is referred to as the Main Waste Dump Wrap. This will shallow the slope at the toe of the main waste dump to a minimum grade of 3H:1V.

The Main Waste Dump Expansion is currently being used as a location to stockpile overburden. Most of reclamation activities, including upgrade of the currently placed cover, remain to be

completed. Re-graded slopes are proposed to be variable, and are currently designed to be as steep as 2.5H:1V (for short transition segments), and as shallow as 4H:1V.

Consistent with the overall theme of this document, a minimum cover thickness of 0.5 m is proposed recognizing that localized areas may require additional grading to meet the final design grades. Much of the dump slopes will require re-grading and, except for the south-east corner of the dump, this will be completed on a balanced cut-fill basis using the existing waste rock.

To shed water, the top of the Main Waste Dump Expansion has been designed to be split into two main catchments with water shed to the west. These catchment areas are illustrated in the detailed hydraulic analysis information contained in Appendix F. The intent is to minimize the amount of water that flows directly over the slope as sheet flow to that which falls and accumulates on the slopes. Each of these small catchments would be constructed to direct water into a broad swale, currently considered to be 2 m wide at the base, and have side slopes of 10H:1V. The swales were designed based on contributing watershed area, and flow depths were calculated to be between 0.14 m and 0.34 m during a 1:200 year 24-hour flood event. The reported depths are double that of the calculated depths to allow for ice accumulation; this is conservative, as explained in Section 5.1 for the Southwest Waste Dump.

The swales have been designed to flow over the slope, and transition to a 2 m wide base channel with 3H:1V side slopes. The swales on the top are proposed to be protected with vegetation, while the slopes are proposed to be armoured with a gravel to cobble sized rip rap. Rip rap thickness and final dimensions are yet to be determined. The proposed re-grading and swale locations are illustrated in Figure 2.

As discussed in Section 3.3, soil loss due to erosion caused by overland sheet flow can impact the integrity of the cover. Due to the size of the dump, and the length of the slopes, a variety of support practices have been adopted to reduce the overall erosion susceptibility of the slope. These support practices include complex slopes, benches to reduce flow velocity and provide areas of sediment deposition. Of critical importance will be the establishment of a strong vegetative cover early in the closure to further reduce the potential for erosion along the slopes. The short-term revegetation concept is described in Section 4.

The proposed plan includes approximately 231,800 m^3 of cover material, considering the additional surface area created with the shallowing of the dump toe (the Main Waste Dump Wrap). It is understood that approximately half of this volume has been hauled and stockpiled at the Main Waste Dump Expansion.

5.8 Main Pit Dump & Subaerial Main Pit Tailings

The Main Pit Dump is currently under construction and the final configuration of the dump has yet to be finalized. SAT material above elevation 786m (the planned final water level of the pit) is currently proposed to be relocated to areas within the pit where it can be covered with water (i.e. below elevation 786 m). Tailings were deposited in the Main Pit both subaqueously and subaerially, and some of these tailings are above the planned final water elevation of 786m, and as such require a cover.

A minimum cover thickness of 0.5 m is proposed over the Main Pit Dump. To shed water, the top of the Main Pit Dump has been designed to shed water towards the south-west. The catchment area anticipated to capture the flow is illustrated in the detailed hydraulic analysis information contained in Appendix F. A broad swale, with dimensions of 2.0 m wide at the base and side slopes of 10H:1V is currently proposed. The swale was designed based on contributing watershed area and the flow depth has been estimated to be 0.22 m during a 1:200 year 24-hour flood event, accounting for ice accumulation within the swale; this is conservative, as explained in Section 5.1 for the Southwest Waste Dump. A swale is also proposed to carry the water collected on the top surface down the slope. This swale has been designed to flow over the slope and transition to a channel with the same base width, but with 3H:1V side slopes. The swale on the top are proposed to be protected with vegetation, while the slopes are proposed to be armoured with a gravel to cobble sized rip rap. Rip rap thickness and final dimensions are yet to be determined.

A cover will be required for the subaerially deposited tailings which are expected to be above the high-water level in the pit at closure. These tailings will be covered with a rock trafficking layer and a 0.5 m thick overburden cover; however, detailed engineering has not been advanced for this concept. The trafficking layer will be designed with the intent of preventing boils from developing through the cover and preventing wave erosion of tailings where water depths are less than 1 m (i.e. above 785 m elevation). As a conservative measure at this point, a 20% increase in the overall area to be covered has been accounted for.

A preliminary plan for re-grading and the swale location is illustrated on Figure 2. The proposed plan considered as balanced cut-fill to re-grade the dump, and approximately 67,800 m^3 of cover material over the dump. The subaerial tailings are anticipated to require approximately 6,400 m^3 of cover material, and 12,800 m^3 of trafficking material.

As discussed in Section 3.3, soil loss due to erosion caused by overland sheet flow can impact the integrity of the cover. However, it is proposed that the revegetation concepts described in Section 4 be implemented on the slopes to minimize erosion, and increase the rate of success for revegetation.

The currently illustrated dump faces are at a minimum slope of 3H:1V, and therefore the physical stability of the cover is not anticipated to be problematic.

5.9 Area 118 Pit Backfill Dump

The Area 118 Pit Backfill Dump has been receiving off-spec overburden from the excavation of the Area 2 Stage 3 Pit. The proposed re-grading plan is intended to fill in the dump and shed water towards the Area 2 Pit, with slopes not steeper than 3H:1V. A minimum cover thickness of 0.5 m is proposed. It is currently proposed that the dump will be graded but not covered with additional material, as a final cover of residuum or fine grained material to match the surrounding hillsides would be appropriate.

The current grading concept separates the dump into three catchment areas (detailed provided in Appendix F). Broad swales with dimensions of 2 m wide at the base and side slopes of 10H:1V is

currently proposed. The swales were designed based on contributing watershed area and the flow depth has been estimated to be between 0.08 m to 0.16 during a 1:200 year 24-hour flood event, accounting for ice accumulation within the swale; this is conservative, as explained in Section 5.1 for the Southwest Waste Dump. The swale is proposed to be protected with vegetation. Where the swales transition along the slopes, the side slopes will transition to 3H:1V and be armoured with gravel to cobble sized riprap. Detailed design of these swales has yet to be completed.

A preliminary plan for re-grading and the swale location is illustrated in Figure 2. The proposed plan includes approximately 26,400 m^3 of cover material.

As discussed in Section 3.3, soil loss due to erosion caused by overland sheet flow will be an important consideration for this facility, and the revegetation concepts described in Section 4 will be incorporated into the design.

The currently illustrated dump faces are at angles of 3H:1V or shallower, and therefore the physical stability of the cover is not considered to be problematic.

6 Conclusions

This document presents an updated design for the closure covers at the Minto site and supersedes any previous cover designs at the site. A description of how the design objectives and function were met is provided in the table below:

Design Objective and Function	Design Component
Minimize infiltration to the extent practical using locally available material.	Specifying a material particle size distribution demonstrated to meet appropriate reduce infiltration through numerical modelling (SRK, 2015).
Ensure a stable landform that will promote establishment of natural vegetation endemic to the area.	Analysis has been completed to demonstrate the erosion susceptibility of the proposed cover soils; however, methods to limit the erosion have been proposed. Analysis has been completed to determine the physical stability of the covers.
Minimize ponding and surface erosion on the final landform.	The grading plans developed have been done so to reduce the risk of surface water ponding, and flow velocities have been considered in completing the sizing of the channels.

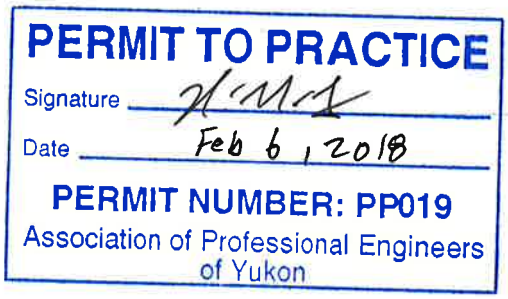
Other primary conclusions of this document include:

- Approximately 2.3 Million bank cubic metres is available for use as closure cover from the Reclamation Overburden Dump.
- The cover thickness is proposed to have greater than 10% fines, and be placed at a minimum 0.5 m thickness. Based on current grading plans, this results in approximately 1,077,300 m^3 of cover material to be placed (in-situ thickness).
- Material meeting the cover specification was direct hauled from the Area 2 Stage 3 pit excavation and stockpiled at the Southwest Waste Dump, Main Waste Dump Expansion, Dry Stack Tailings Storage Facility, and Mill Valley Fill Extension.
- Revegetation seed mixes have been developed for sloped areas and level areas. These seed mixes have been purchased, and the seed is currently on-site.
- Facilities have been designed with re-graded slopes intended to provide a cut/fill balance while targeting overall slopes as shallow as possible.
- Facilities are to be graded to reduce the amount of water that flows over slopes. Facility tops are graded to central swales designed to accommodate a 1:200 year event, without the need for aggregate riprap, and that rely on vegetation to provide roughness within the channel.
- Additional work is required prior to finalizing the implementation of these cover designs.

This report, Updated Closure Cover Design for the Minto Mine 2018 Reclamation and Closure Plan, was prepared by



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All data used as source material plus the text, tables, figures, and attachments of this document have been reviewed and prepared in accordance with generally accepted professional engineering and environmental practices.

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The opinions expressed in this report have been based on the information available to SRK at the time of preparation. SRK has exercised all due care in reviewing information supplied by others for use on this project. Whilst SRK has compared key supplied data with expected values, the accuracy of the results and conclusions from the review are entirely reliant on the accuracy and completeness of the supplied data. SRK does not accept responsibility for any errors or omissions in the supplied information, except to the extent that SRK was hired to verify the data.

7 References

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Figures



REFERENCE

1. Ortohophoto date August 2017 and provided by Minto.
2. This figure should be read in conjunction with the Updated Closure Cover Design for the Minto Mine 2018 Reclamation and Closure Plan Report. Dated January 2018.

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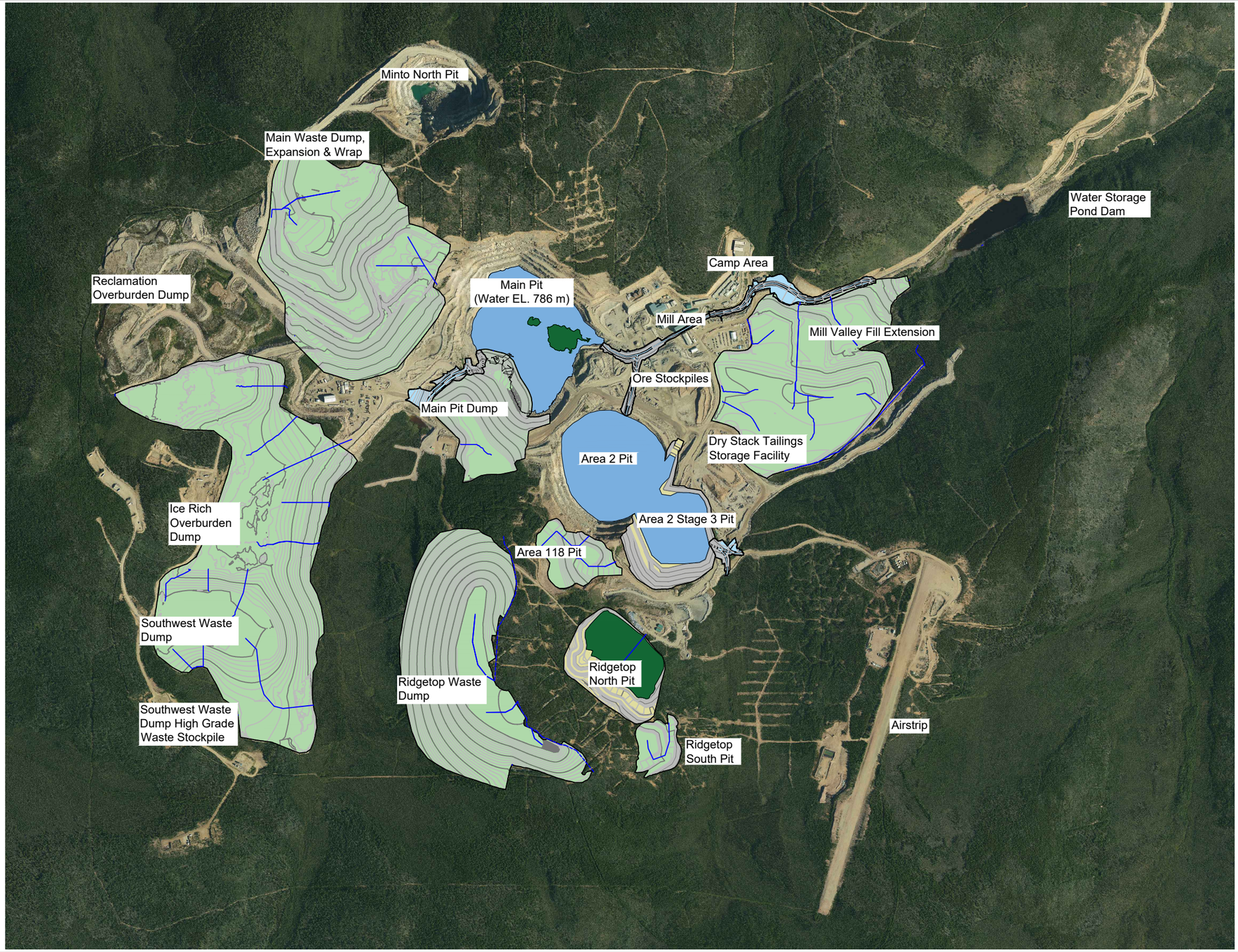
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2018 Updated Closure Cover Design

Existing Site Conditions

DATE:	APPROVED:	FIGURE:
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LEGEND

- Subaerial Tailings Cover (Detail 1, Figure 3)
- Cover Application Area (Detail 2, Figure 3)
- Open Pit Slopes
- Water
- Primary Water Conveyance Feature
- Secondary/Tertiary Water Conveyance Feature (Details 3 & 4, Figure 3)

REFERENCE

1. Contours shown as 2.0 m intervals
2. Ortohophoto date August 2017 and provided by Minto.
3. This figure should be read in conjunction with the Updated Closure Cover Design for the Minto Mine 2018 Reclamation and Closure Plan Report. Dated January 2018.

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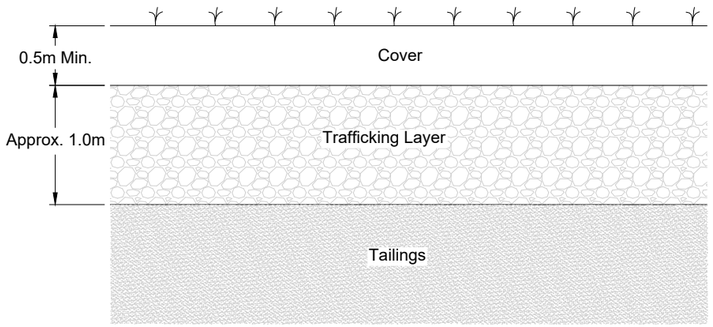
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2018 Updated Closure Cover Design

Final Closure Configuration

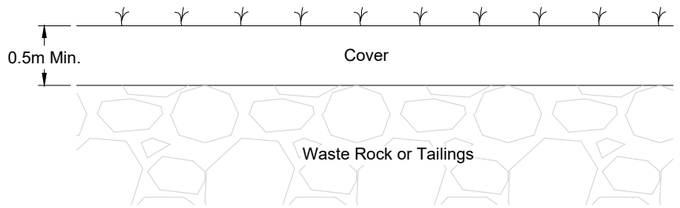
DATE: January 2018 | APPROVED: EPK | FIGURE: 02

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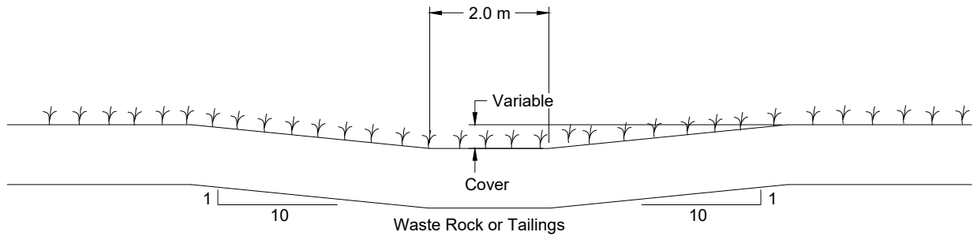
1 Subaerial Tailings Cover

Not To Scale



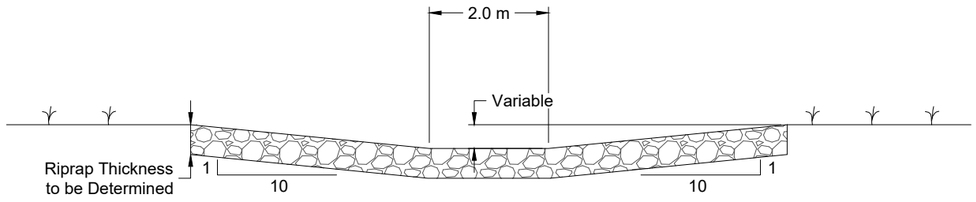
2 Cover Detail

Not To Scale



3 Swale Detail

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4 Slope Swale Detail

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SRK JOB NO.: 1CM002.049

FILE NAME: 1CM002.049_Details_R1.dwg

Capstone MINTO MINE

OPERATED BY MINTO EXPLORATIONS LTD.

Minto Explorations Ltd.

2018 Updated Closure Cover Design

Details

DATE:	APPROVED:	FIGURE:
January 2018	EPK	03

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Appendix A – Minto Closure Cover Design – Geotechnical Characteristics of the Cover Materials

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Memo

To:	File	**Client:**	Minto Exploration Ltd.
From:	Erik Ketilson, MEng, PEng.	**Project No:**	1CM002.049
Reviewed by:	Maritz Rykaart, PhD, PEng.	**Date:**	July 20, 2016
Subject:	Minto Closure Cover Design – Geotechnical Characterization of the Cover Materials		

1 Introduction

The Minto Mine requires closure covers to be placed over the waste rock and tailings facilities. The purpose of this memo is to characterize the overburden material to provide an appropriate material envelope to guide the construction of the cover material.

2 Cover Requirements

SRK completed numerical modelling to bracket percolation predictions in 2015 (SRK, 2015). The modelling indicated that without a cover, the net percolation could be between 39% and 45% and, depending on climactic conditions, could vary between 26% and 65%. Cover materials were applied, with the base case cover material consisting of material properties from sample MWD-TP4 collected at the Minto mine, which is a gravel and sand material with more than 25% fines (<0.075 microns). The results indicate that the net percolation is approximately 23%, but depending on the climactic conditions, can vary from 6% to 43%. The cover material was varied and an analysis was also completed considering a coarse material (MWD-TP3), which is predominantly gravel and sand with less than 10% fines (<0.075 microns). The results of this analysis indicated that the net percolation was approximately 23%, but depending on the climactic conditions, can vary from 5% to 44%.

The modelling results indicate that there is nearly a 20% decrease in net percolation after cover materials are included in the analysis. The cover thickness was assumed to be approximately 0.5 m thick, and following sensitivity analysis to the thickness (1 m and 2 m) minor decreases in net percolation were estimated. Therefore, a minimum cover thickness of 0.5 m was adopted for the project, with materials containing greater than 10% fines. Based on preliminary closure cover revegetation work completed by Integral Ecology Group (IEG) in 2016 (IEG, 2016), overburden materials containing greater than 10% fines are expected to support the growth of vegetation.

3 Geotechnical Material Characteristics

3.1 Particle Size Distribution

SRK evaluated 167 particle size distribution analyses completed on overburden samples at the Minto Site, and classified the particle size distribution based on the Modified Unified Soil Classification System (MUSCS) and the Unified Soil Classification System (USCS). Data was obtained from investigations completed, and documented by SRK (SRK, 2013, 2016a&b). By correlating the specific sample analysis with borehole location, SRK was able to classify the overburden by source location. Figure 1 illustrates the upper and lower bound of the particle size distributions for each area.

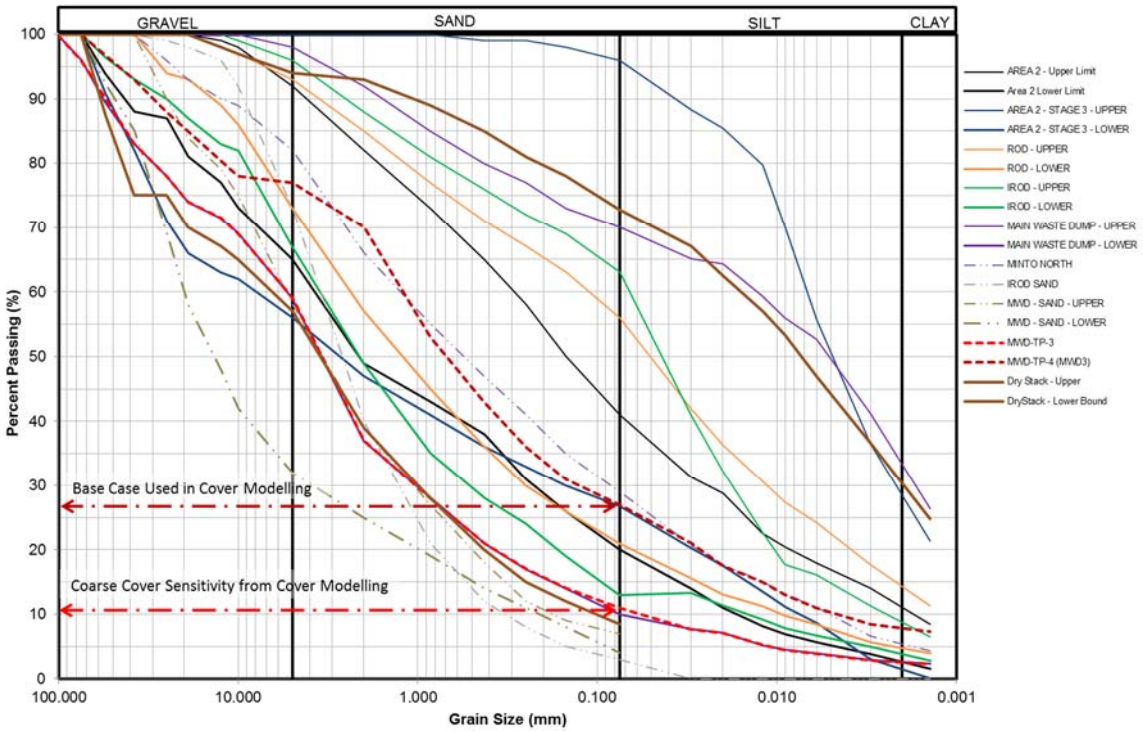


Figure 1: Particle Size Distribution

Source: Minto_MaterialProperties_1CM002-049_Rev00_EK.xlsx

The upper boundaries are also listed in tabular form in Table 1. The lower boundaries are listed in tabular form in Table 2.

Table 1: PSD Area Upper Bound

Upper Bound - Finer Limit					
Area	**Gravel**	**Sand**	**Fines**	**Silt**	**Clay**
Area 2	8	51	41	31	11
Area 2 - Stage 3	0	4	96	69	27
Drystack Tailings Cover	6	21	73	44	29
Ice Rich Overburden Dump	4	33	63	55	8
Main Waste Dump	2	28	70	38	32
Reclamation Overburden Dump	7	37	56	42	14

Source: Minto_MaterialProperties_1CM002-049_Rev00_EK.xlsx

Table 2: PSD Area Lower Bound

Lower Bound - Coarser Limit					
Area	**Gravel**	**Sand**	**Fines**	**Silt**	**Clay**
Area 2	35	45	20	18	2
Area 2 - Stage 3	44	29	27	26	1
Drystack Tailings Cover	43	49	9	2	6
Ice Rich Overburden Dump	33	54	13	9	4
Main Waste Dump	41	49	10	7	3
Reclamation Overburden Dump	27	52	21	16	5

Source: Minto_MaterialProperties_1CM002-049_Rev00_EK.xlsx

3.2 Atterberg Limits

Of the available samples, 38 had completed analysis to determine the Atterberg Limits (liquid and plastic limits). Data was obtained from investigations completed, and documented by SRK (SRK, 2013, 2016a&b). Figure 2 illustrates soil classification according to the modified unified soil classification system. The modified unified soil classification system is similar to the Unified Soil Classification System, however, splits low plastic clay classification into two categories including a clay of intermediate plasticity (CI).

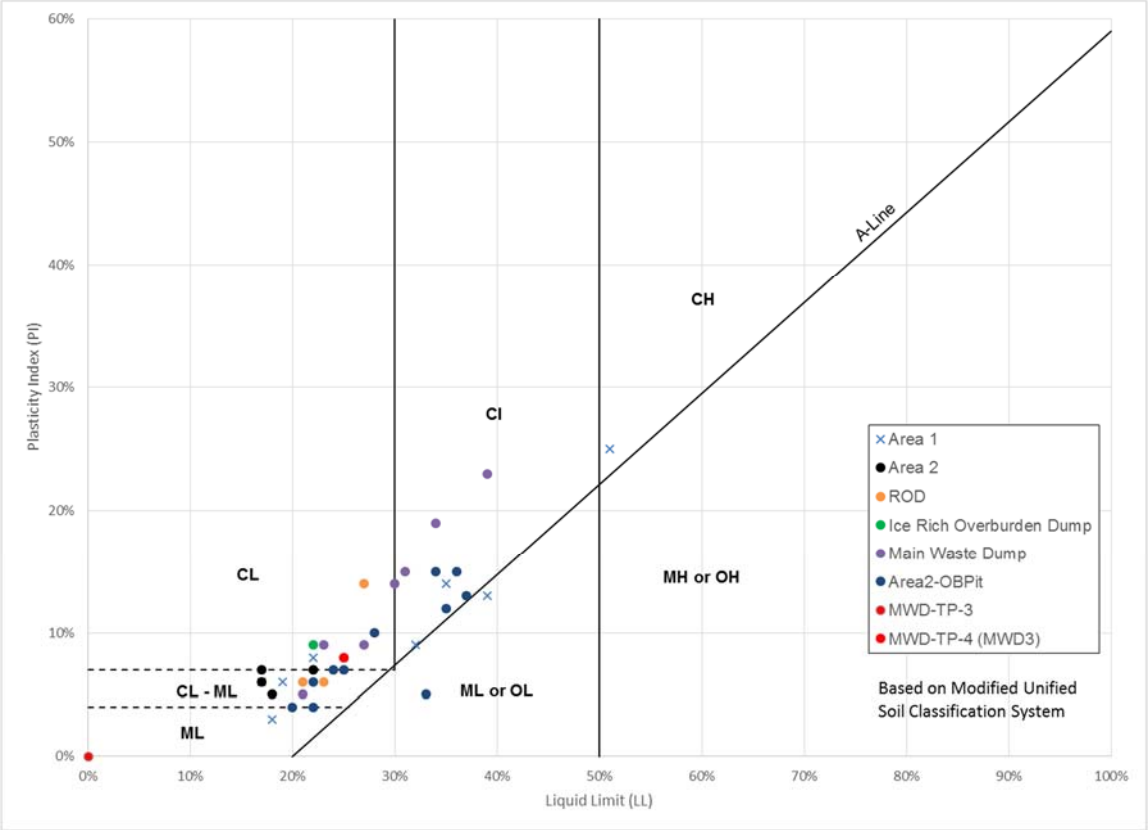


Figure 2: Atterberg Limits

Source: Minto_MaterialProperties_1CM002-049_Rev00_EK.xlsx

3.3 Strength

Strength characteristics for the site materials at Minto are provided within the review of geotechnical strength properties memorandum prepared by SRK (SRK, 2014).

3.4 Erosion

Soil erosion classification is based on the USDA soil textural classification. Table 3 provides a summary of the particle size diameter range based on the Unified soil classification system (USCS), and the United States Department of Agriculture (USDA) soil textural classification.

Table 3: USCS vs USDA Particle Size Distribution Systems

Soil Component	Particle Size Diameter Rage (millimeters)	
	USCS / MUSCS	USDA
Boulders	> 200	
Cobbles	200 – 76	
Gravel	76 – 4.75	
Sand	4.75 – 0.075	2.0 – 0.05
Silt	0.075 – 0.002	0.05 – 0.002
Clay	< 0.002	< 0.002

The samples were re-classified, and plotted on a soil texture triangle (Figure 3) to determine the general soil texture. The samples typically categorized as sandy loam.

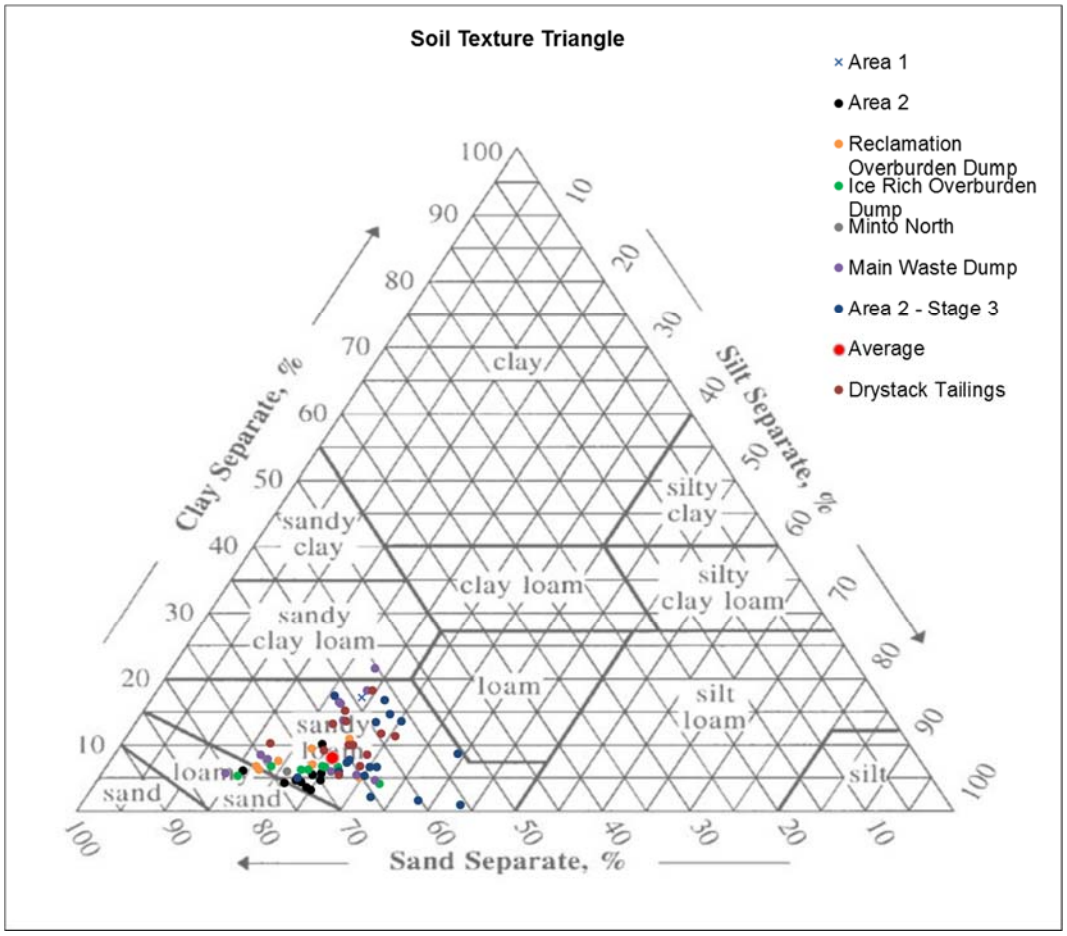


Figure 3: Soil Classification Based on Soil Texture Triangle

Source: Minto_MaterialProperties_1CM002-049_Rev00_EK.xlsx

4 Design Parameters

Based on the particle size distributions, and the net percolation cover modelling completed (SRK, 2015), Table 1 provides a summary of proposed design parameters to be adopted for the Minto Closure Covers.

Table 4: Closure Cover Design Parameters

Description		Value
Cover Thickness		0.5 m (minimum)
Cover Material Specifications	Gravel	0 % to 40%
	Sand	60% to 90%
	Fines	> 10%
Soil Texture Classification	Sandy Loam	Sandy Loam

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The opinions expressed in this report have been based on the information available to SRK at the time of preparation. SRK has exercised all due care in reviewing information supplied by others for use on this project. Whilst SRK has compared key supplied data with expected values, the accuracy of the results and conclusions from the review are entirely reliant on the accuracy and completeness of the supplied data. SRK does not accept responsibility for any errors or omissions in the supplied information, except to the extent that SRK was hired to verify the data.

5 References

(IEG, 2016). Minto Mine preliminary closure cover revegetation plan. Memorandum. Prepared for SRK Consulting (Canada) Inc. for the Minto Mine Site. Integral Ecology Group. July 29, 2016.

(SRK, 2013). 2012 Overburden Characterization Data Report for Minto Closure Covers. Report. Prepared for Minto Explorations Ltd. SRK Consulting (Canada) Inc. Project No. 1CM002.007. October 2013.

(SRK, 2014). Review of geotechnical strength properties at Minto Mine. Technical Memorandum. Prepared for Minto Explorations Ltd. SRK Consulting (Canada) Inc. Project No. 1CM002.018.110. July 17, 2014.

(SRK, 2015). Minto Mine Closure Covers: Results of Numerical Modelling to Bracket Percolation Predications. Technical Memorandum. Prepared for Minto Explorations Ltd. SRK Consulting (Canada) Inc. Project No. 1CM002.030. January 5, 2015.

(SRK, 2016a). Dry-stack Tailings Storage Facility Interim Cover Investigation. Technical Memorandum. Prepared for Minto Explorations Ltd. SRK Consulting (Canada) Inc. Project No. 1CM002.031.0500.03. April 7, 2016.

(SRK, 2016b). Minto Mine – 2014 and 2015 Area 2 Stage 3 Overburden Drilling. Technical Memorandum. SRK Consulting (Canada) Inc. Project No. 1CM002.037. July 2016.

Appendix B – Minto Mine – 2014 and 2015 Area 2 Stage 3 Overburden Drilling

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Memo

To:	File	**Client:**	Minto Explorations Ltd.
From:	Kaitlyn Kooy, EIT; Erik Ketilson, MEng, PEng	**Project No:**	1CM002.037
Reviewed by:	Maritz Rykaart, PhD, PEng	**Date:**	July 13, 2016
Subject:	Minto Mine - 2014 and 2015 Area 2 Stage 3 Overburden Drilling		

1 Introduction

The Minto Mine is a high-grade copper mine located in the Yukon, approximately 240 km north of Whitehorse. The mine site occupies the valley in the upper reaches of Minto Creek, a tributary on the west side of the Yukon River, about 9 km from the mouth. Operations are ongoing at this time (2016) and began in October 2007. Three pits have been completed to date: the Main Pit, the Area 118 Pit, and the Area 2 Stage 2 Pit.

In support of developing a more robust understanding of the resource within the proposed Stage 3 expansion of the Area 2 pit, Minto planned and executed a drilling program. This program provided an opportunity for Minto to evaluate if there may be minable units of cover material within the overburden that will be excavated as part of the pit development. SRK recommended that the Minto staff receive some basic training in soil logging to help identify possible soils that could be beneficial for closure purposes. SRK provided Minto with on-site staff training for soil logging of some boreholes completed in 2014 and 2015.

The memorandum provides a summary of SRK's assistance during the drilling program, interpretation of the material properties, and comments regarding the potential for mineable units of cover material within the identified overburden for boreholes completed under the supervision of SRK.

2 Field Visit

SRK's Murray McGregor, EIT, visited the Minto site from October 27th through November 5th 2014. During this time, Mr. McGregor was responsible for logging the overburden soils from the initial drill holes as well as training several site staff in basic soil logging methods. Soils were logged according to the Unified Soil Classification System and samples of each material unit were collected for laboratory testing (discussed in Section 4.0). Boreholes were advanced using an HQ diameter, diamond drill bit.

3 Borehole Locations

Table 1 provides a list of boreholes drilled during the 2014 and 2015 programs. The locations of the boreholes are provided in Figure 1. Borehole logs are provided in Attachment 1 and photos of the core form Attachment 2.

Table 1: 2014 – 2015 Drilling Program Boreholes

Hole ID	Easting(1)	Northing(1)	Drill Program
14-SWC-966	385107.8	6944363.1	Oct/Nov 2014
14-SWC-967	385146.1	6944317.3	Oct/Nov 2014
14-SWC-968	385169.1	6944289.2	Oct/Nov 2014
14-SWC-969	385172.2	6944225.0	Oct/Nov 2014
14-SWC-970	385125.3	6944272.0	Oct/Nov 2014
14-SWC-971	385075.1	6944288.0	Oct/Nov 2014
14-SWC-972	385215.6	6944284.0	Oct/Nov 2014
14-SWC-973	385164.4	6944349.0	Oct/Nov 2014
14-SWC-974	385140.0	6944379.0	Oct/Nov 2014
14-SWC-975	385156.1	6944428.0	Oct/Nov 2014
14-SWC-976	385075.3	6944328.0	Oct/Nov 2014
14-SWC-977	385097.9	6944254.0	Oct/Nov 2014
14-SWC-978	385122.4	6944219.0	Oct/Nov 2014
14-SWC-979	385218.8	6944233.0	Oct/Nov 2014
14-SWC-980	385194.5	6944261.0	Oct/Nov 2014
14-SWC-981	385148.0	6944188.0	Oct/Nov 2014
15-SWC-995	385096.1	6944251.6	Feb 2015
15-SWC-996	385161.0	6944314.9	Feb 2015
15-SWC-997	385140.9	6944124.0	Feb 2015

Notes:

(1) Easting and Northing presented in NAD 1983 UTM UTM Zone 8N

4 Soil Description

The encountered overburden materials within the Area 2, Stage 3 pit consists of layers with variable thickness of sand, gravel, silts, and clays. The borehole logs are included as Attachment 1, and photos of the core collected form Attachment 2.

An organic layer was noted only in one borehole – 14-SWC-979. The layer was identified between 0.7 m and 1.0 m below ground surface, and the layer was identified to contain primarily clay and silt sized particles with trace sand sized particles encountered. The soil was brown/black in colour, moist, and of medium plasticity.

Discontinuous layers of sand were identified in each of the boreholes. The sand was quite variable, and consisted of varying content of gravel, silt, and clay content and ranged from well to poorly graded, based on visual identification. The sand was generally brown, with varying shades of grey and red throughout. The sand was primarily identified as moist, with the exception of discreet layers in 14-SWC-968, 14-SWC-970, 14-SWC-971, 14-SWC-972, 14-SWC-973, 15-SWC-995, and 15-SWC-996 where the sand was identified as wet. In some cases, the sand was classified with low or medium plasticity – this was identified in boreholes 14-SWC-967, 14-SWC-969, 15-SWC-995, 15-SWC-996, and 15-SWC-997. The sand was otherwise identified as non-plastic. Generally, the sand unit was considered massive with the exception of 14-SWC-967; 15-SWC-997, and 15-SWC-997 where layers of sand were identified to be laminated or blocky.

Gravel layers were identified in boreholes 14-SWC-966; 14-SWC-967; 14-SWC-973; 14-SWC-975; 14-SWC-981; and 15-SWC-997. The gravel contained varying levels of sand, silt, and clay content. The material was generally brown, with varying shades or red and grey. The material was primarily identified as moist, with the exception of some layers in 15-SWC-997 which were identified as wet. The gravel layers were non-plastic, and massive.

Material classified as silt contained varying degrees of clay, sand, and gravel. The material was generally classified as poorly graded; however, layers were identified in 14-SWC-967, 14-SWC-972, 14-SWC-973, 15-SWC-995, 15-SWC-996, and 15-SWC-997. Generally, the silt was identified to be brown, with varying shades of grey. The moisture content was visually identified as primarily moist, with the exception of some layers identified in borehole 14-SWC-972, 14-SWC-974, 14-SWC-979, 15-SWC-995, and 15-SWC-996 as wet; and one layer from 28.6 m to 28.9 m in borehole 14-SWC-979 was identified as dry. The silt ranged from low to high plasticity. The material was generally classified as massive, with laminated layers identified in boreholes 14-SWC-969, 14-SWC-979, 15-SWC-996, and 15-SWC-997; and one layer was identified as blocky in each of boreholes 14-SWC-973 and 15-SWC-996.

Clay layers primarily contained silt with varying level of sand and gravel. The clay was generally poorly graded, and grey in colour, although some layers were identified as brown. The material was moist, with the exception of layers identified in 14-SWC-974 and 15-SWC-995 which were identified as wet. Clay samples were generally exhibited medium to high degrees of plasticity.

Bedrock was encountered at depths ranging between 3.15 m (14-SWC-966) to 26.5 m (14-SWC-973).

5 Laboratory Testing

During a subsequent site visit in February 2015, Mr. McGregor selected samples from the combined drilling of 2014 and 2015 programs to seek out continuous units of fine grained material containing clay which could be used in cover construction. Several samples were collected from boreholes beneath the Area 2 ring road. A summary of the laboratory tests performed are shown in Table 2.

Table 2: Laboratory Tests Performed

Hole ID	From (m)	To (m)	Sample ID	Moisture Content	Particle Size Analysis	Atterberg Limits
14-SWC-968	15.05	15.40	MM-101259	✓	✓	✓
14-SWC-970	5.65	6.00	MM-101262	✓	✓	✓
14-SWC-969	12.00	12.30	MM-101270	✓	✓	
	19.05	19.35	MM-101273	✓	✓	
14-SWC-972	9.25	9.55	MM-101277	✓	✓	✓
	18.55	18.85	MM-101280	✓	✓	
14-SWC-973	2.80	3.10	MM-101285	✓	✓	✓
	9.17	9.47	MM-101288	✓	✓	✓
14-SWC-975	5.70	6.00	MM-101298	✓	✓	✓
14-SWC-979	20.60	20.90	MM-101317	✓	✓	✓
14-SWC-980	19.40	19.70	MM-101326	✓	✓	✓
15-SWC-995	8.60	8.90	58554	✓	✓	
15-SWC-996	6.12	6.42	58557	✓	✓	✓
	10.69	11.00	58558	✓		
	14.73	15.10	58559	✓	✓	✓
15-SWC-997	9.79	10.11	160025	✓	✓	✓
	15.64	15.88	58565	✓		
	23.09	23.43	58568	✓	✓	✓
	30.60	30.91	58570	✓		
	41.32	41.67	58572	✓	✓	✓

5.1 Test Results

The test results are presented in Attachment 3. Natural moisture content analysis results are summarized in Table 3. Particle size distribution analysis is illustrated in Figure 1, while results of the Atterberg Limits, as classified using the Modified Unified Soil Classification System (MUSCS), is illustrated in Figure 2.

Table 3: Natural Moisture Content Analysis Results

Hole ID	From (m)	To (m)	Moisture Content
14-SWC-968	15.05	15.40	16.6%
14-SWC-970	5.65	6.00	27.6%
14-SWC-969	12.00	12.30	10.7%
	19.05	19.35	26.3%
14-SWC-972	9.25	9.55	16.6%
	18.55	18.85	31.7%
14-SWC-973	2.80	3.10	35.9%
	9.17	9.47	13.2%
14-SWC-975	5.70	6.00	28.1%
14-SWC-979	20.60	20.90	26.5%
14-SWC-980	19.40	19.70	18.8%
15-SWC-995	8.60	8.90	13.4%
15-SWC-996	6.12	6.42	14.8%
	10.69	11.00	21.7%
	14.73	15.10	16.1%
15-SWC-997	9.79	10.11	10.3%
	15.64	15.88	13.7%
	23.09	23.43	11.7%
	30.60	30.91	15.4%
	41.32	41.67	12.7%

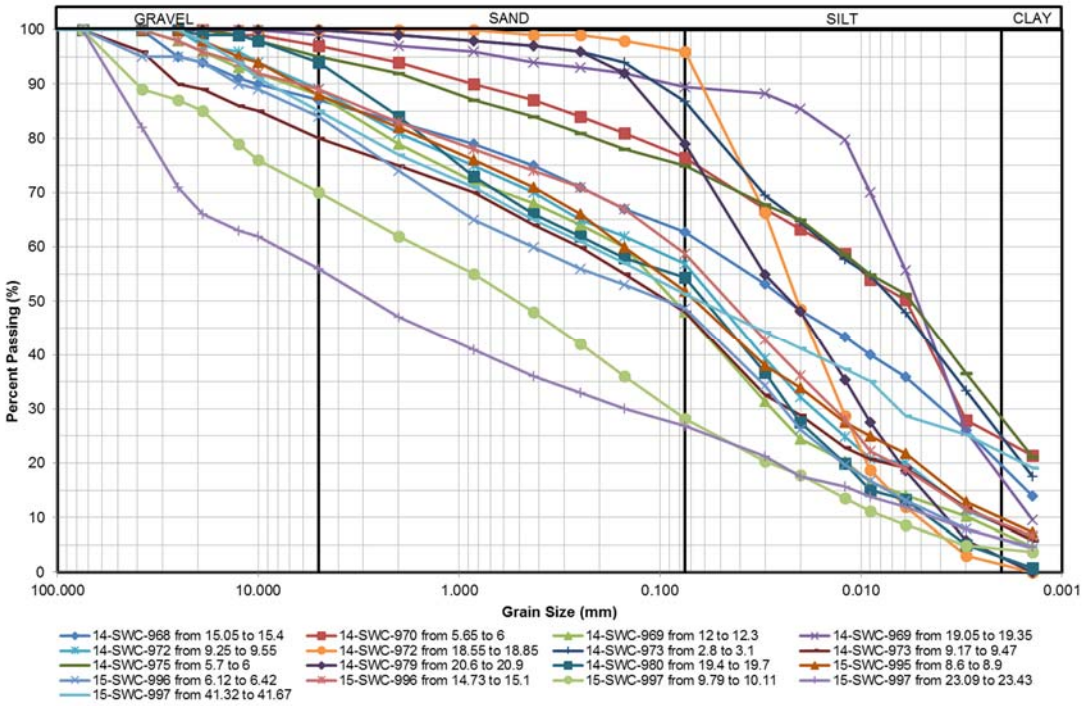


Figure 1: Particle Size Distribution Analysis Results

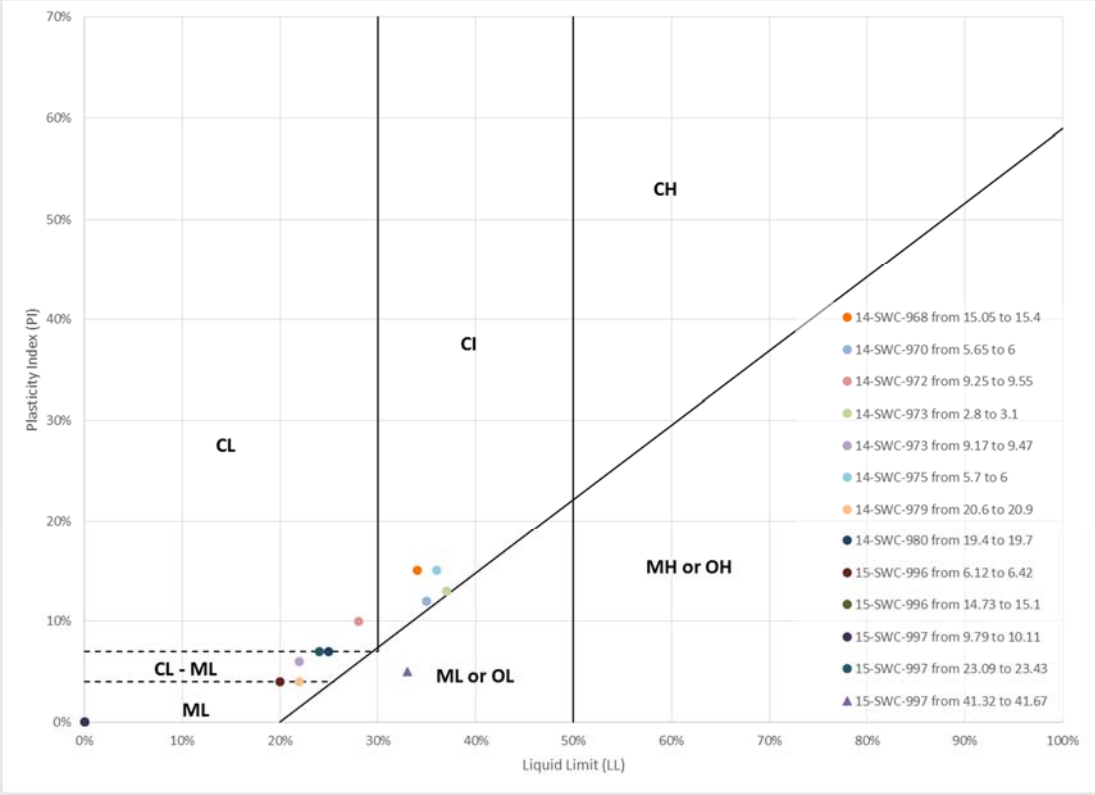


Figure 2: Atterberg Limit Analysis Results

6 Conclusions

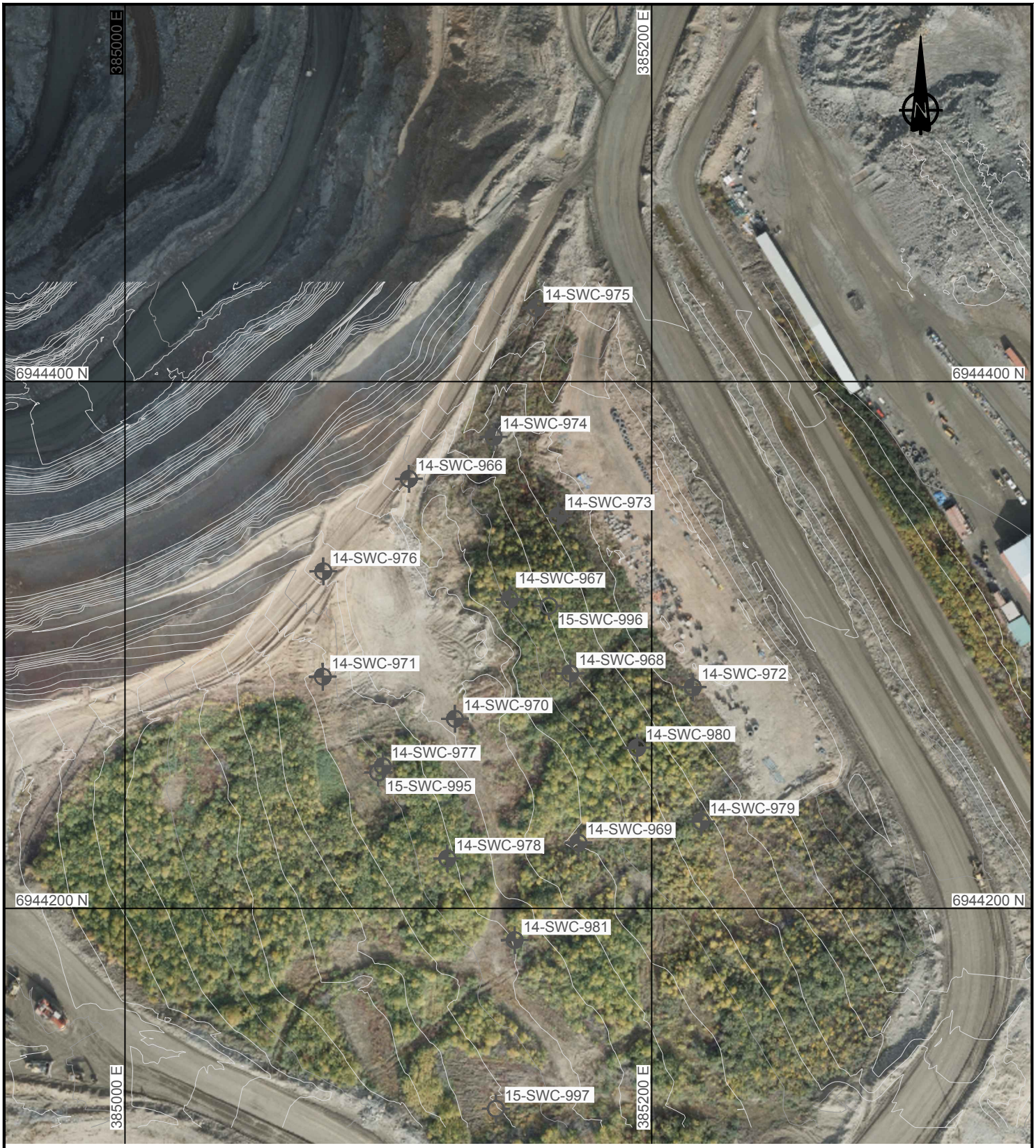
Based on review of the borehole logs, and laboratory testing results, the data suggests that the overburden material encountered is not present in appropriate continuity to facilitate effective selection and segregation of materials using mass mining methods.

Testing indicates that the natural moisture content ranges between 10% and 36%. Particle size distributions analysis indicated fines content ranging between 26% and 96%; with clay sized particles ranging between <1% to approximately 30%. Atterberg limit analysis indicated that the sample classification was variable, including intermediate plasticity clays, and low plasticity clays and/or silts.

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Figure



LEGEND

- Oct./Nov. 2014 Drill Program
- Feb. 2015 Drill Program

NOTES

1. Data presented in NAD83 UTM Zone 8N.
2. Orthophoto provided by Minto Mine, 2014.

0 20 40 60 80 100
Scale in Metres

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Area 2 Stage 3 Overburden Drilling

Area 2 Stage 3 2014/2015 Overburden Drilling Locations

SRK JOB NO.: 1CM002.037
FILE NAME: OVB_Drilling_Borehole_Plan_1CM002_037.dwg

Minto Mine

DATE:	APPROVED:	FIGURE:
July 2016	KNK	1

Z:\01_SITES\Minto\040_AutoCAD\OVB Drilling 2014-15\OVB_Drilling_Borehole_Plan_1CM002_037.dwg

Attachment 1: Borehole Logs

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Capstone MINTO MINE

PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-966**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385107.8 E 6944363.1 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	830.7
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	827.05
TOTAL DEPTH (m):	3.65

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%)

PSD — Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery %	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
0	830.6				SAND, few gravel, well graded, light brown, sub-angular, moist, non-plastic, non-cohesive, massive	
			MM-101251		BOULDER	Granoriorite
					SAND, few gravel, well graded, light brown, sub-angular, moist, non-plastic, non-cohesive, massive	
2	828.6		MM-101252		GRAVEL, some sand , well graded, light brown, sub-angular, moist, non-plastic, non-cohesive, massive	Minor fines near bedrock transition
					BEDROCK	Weathered granodiorite
4						

srk consulting

capstone MINTO MINE

PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-967**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385146.1 E 6944317.3 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	828.4
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	814.4
TOTAL DEPTH (m):	14

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%) 0 20 40 60 80 100

PSD Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery % (0 20 40 60 80 100)	Sample Type (Grab, SPT)	USCS	Soil Description	Drilling Notes & Additional Comments
0	828				SILT, some sand , little gravel, well graded, black, sub-angular, moist, non-plastic, cohesive, massive	Unclear how much organics exist due to poor recovery
					CLAY, some silt, few sand , few gravel, well graded, grey-brown, sub-rounded, moist, low-plasticity, cohesive, massive	Matrix supported
2	826		MM-1012 53			
						Decreased gravel content
4	824				SAND, some silt, trace gravel, poorly graded, grey-brown, sub-rounded, moist, low-plasticity, cohesive, laminated	Sand fraction is primarily fine
6	822		MM-1012 54			
8	820		MM-1012 55		GRAVEL, few sand , few silt, well graded, brown, sub-angular, moist, low-plasticity, cohesive, massive	Poor recovery from 9m to 10.5m; likely the same material unit with fines washed out
10	818				SAND, little gravel, well graded, light brown, sub-angular, moist, non-plastic, non-cohesive, massive	Primarily medium to coarse sand
			MM-1012 56			
12	816					

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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-967**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385146.1 E 6944317.3 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	828.4
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	814.4
TOTAL DEPTH (m):	14

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%) 0 20 40 60 80 100

PSD Gravel / Sand / Silt / Clay(%)

Depth (m) | Elev (m) | Recovery % 0 20 40 60 80 100 | **Sample Type** Grab SPT

Depth (m)	Elev (m)	USCS	Soil Description	Drilling Notes & Additional Comments
14			BEDROCK	Weathered granodiorite

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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-968**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385169.1 E 6944289.2 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	827.7
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	807.7
TOTAL DEPTH (m):	20

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%)

PSD Gravel / Sand / Silt / Clay(%)

Depth (m) | Elev (m) | Recovery % | Sample Type (Grab, SPT)

USCS	Soil Description	Drilling Notes & Additional Comments
	SAND, few silt, little gravel, well graded, grey-brown, moist, non-plastic, cohesive, massive	Extremely poor recovery in first four runs assumed to be the same material type as observed in run starting at 6m; some matrix
	SAND, some silt, trace gravel, poorly graded, brown, wet, cohesive, massive	

Sample: MM-1012 57

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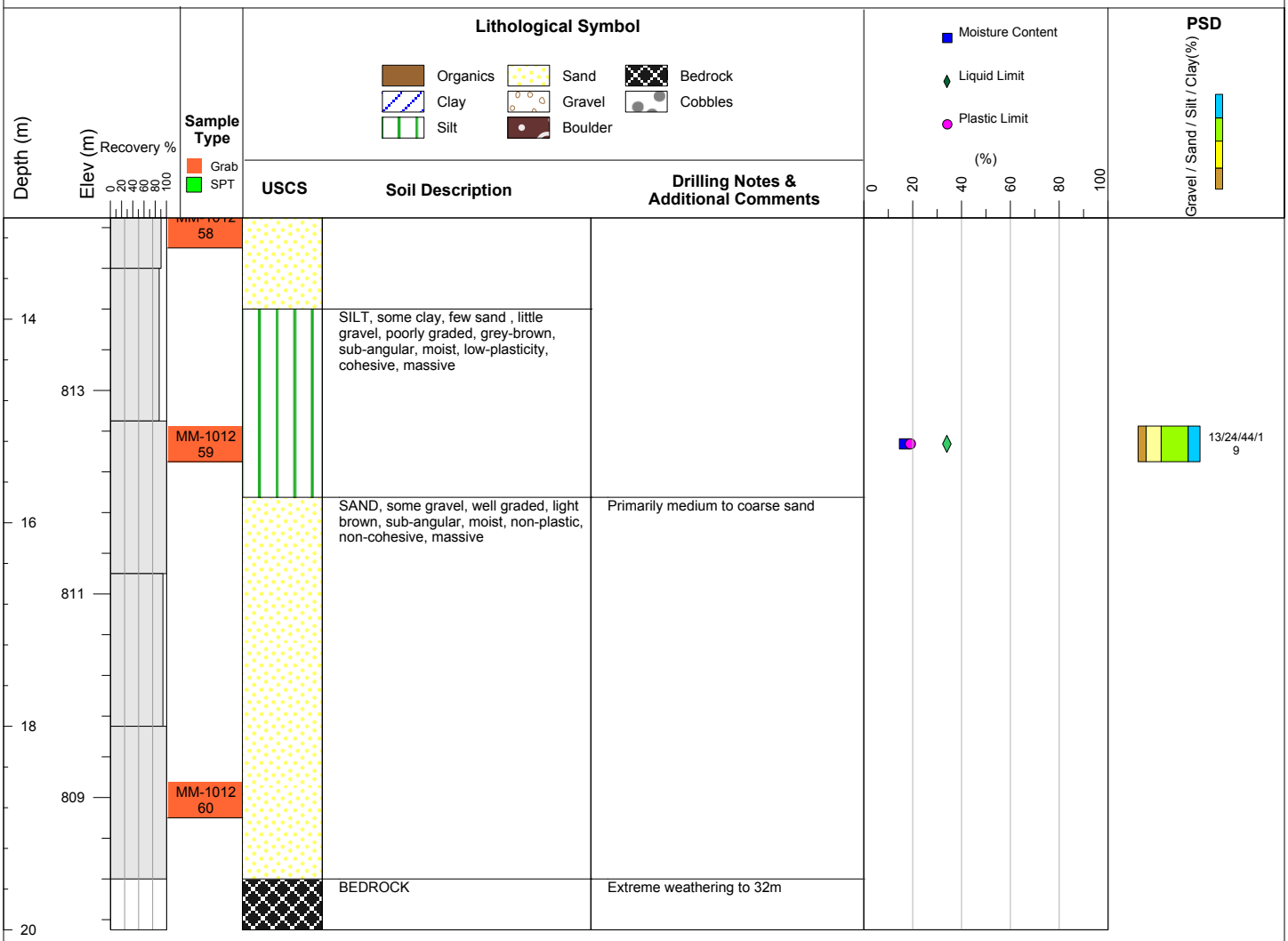
PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-968**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385169.1 E 6944289.2 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	827.7
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	807.7
TOTAL DEPTH (m):	20

DEPOSIT AREA: Area 2



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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-969**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385172.2 E 6944225 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	831.1
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	808.1
TOTAL DEPTH (m):	23

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%) 0 20 40 60 80 100

PSD Gravel / Sand / Silt / Clay(%)

Depth (m) | Elev (m) | Recovery % 0 20 40 60 80 100 | Sample Type: Grab, SPT

USCS	Soil Description	Drilling Notes & Additional Comments
	SAND, few gravel, little silt, well graded, light brown, sub-rounded, moist, non-plastic, non-cohesive, massive	Majority of matrix material (sand) washed out
	SAND, some clay and silt, little gravel, poorly graded, dark brown, sub-angular, moist, low-plasticity, cohesive, massive	
	SILT, some clay, little gravel, poorly graded, dark grey, sub-angular, moist, mid-plasticity, cohesive, laminated	Trace sand
	SAND, some gravel, little clay and silt, well graded, grey brown, sub-angular, moist, non-plastic, cohesive, massive	Occasional cobbles
	SAND, few clay and silt, little gravel, well graded, brown grey, sub-angular, moist, low-plasticity, cohesive, massive	Variable concentrations of silt and silt with clay; coarse brown sand layer at 11.05-11.3m
	SAND, some clay, few silt, poorly graded, dark grey, sub-angular, moist, mid-plasticity, cohesive, massive	Minor organic intervals of <3cm

Depth 0, 2, 4, 6, 8, 10, 12 m — Elev 830, 828, 826, 824, 822, 820 m

MM-1012 68

MM-1012 69

MM-1012 70

12/40/41/7

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HOLE ID:	**14-SWC-969**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014
COORDINATES:	385172.2 E 6944225 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	831.1
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	808.1
TOTAL DEPTH (m):	23

PROJECT: Minto

CLIENT: Minto Exploration Ltd.

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%)

PSD — Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery %	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
0	818		MM-1012 71			
14					SAND, few gravel, few clay and silt, well graded, grey, sub-angular, moist, low-plasticity, cohesive, massive	Occasional cobbles
	816				SAND, few gravel, few silt, well graded, brown grey and reddish grey, sub-angular, moist, non-plastic, cohesive, massive	Minor low-plasticity intervals
16			MM-1012 72			
	814					
18						
	812		MM-1012 73		SILT, some sand , little gravel, poorly graded, grey brown, sub-angular, moist, non-plastic, cohesive, massive	
20					SAND, few silt, little gravel, well graded, reddish grey brown, sub-angular, moist, non-plastic, cohesive, massive	Contact with bedrock
	810					
22			MM-1012 74			
	808				BEDROCK	Extremely weathered to 23.0m, heavily weathered to 29.50 m
24						
	806					

1/10/74/15

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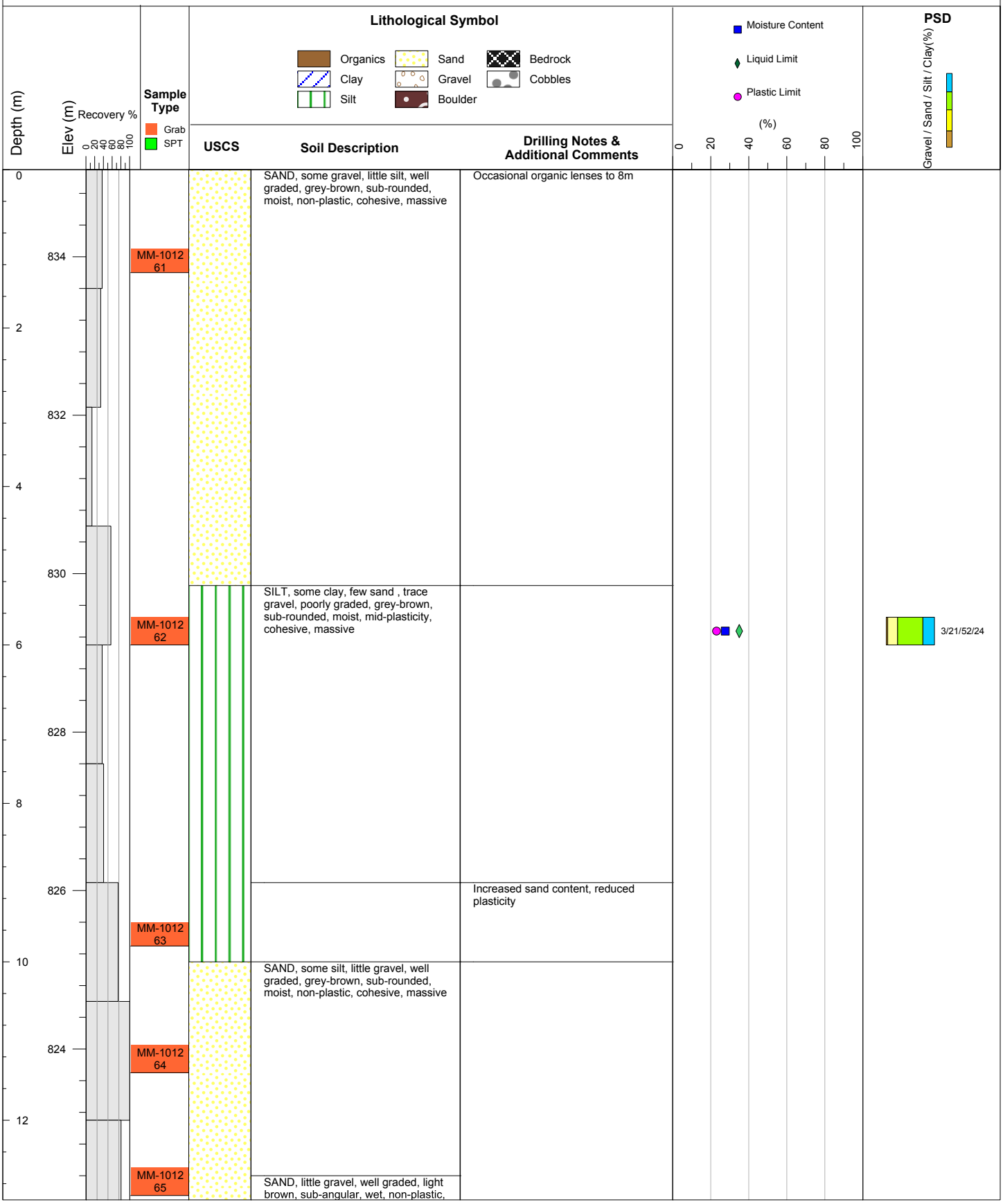
PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-970**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385125.3 E 6944272 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	835.1
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	820.75
TOTAL DEPTH (m):	14.35

DEPOSIT AREA: Area 2



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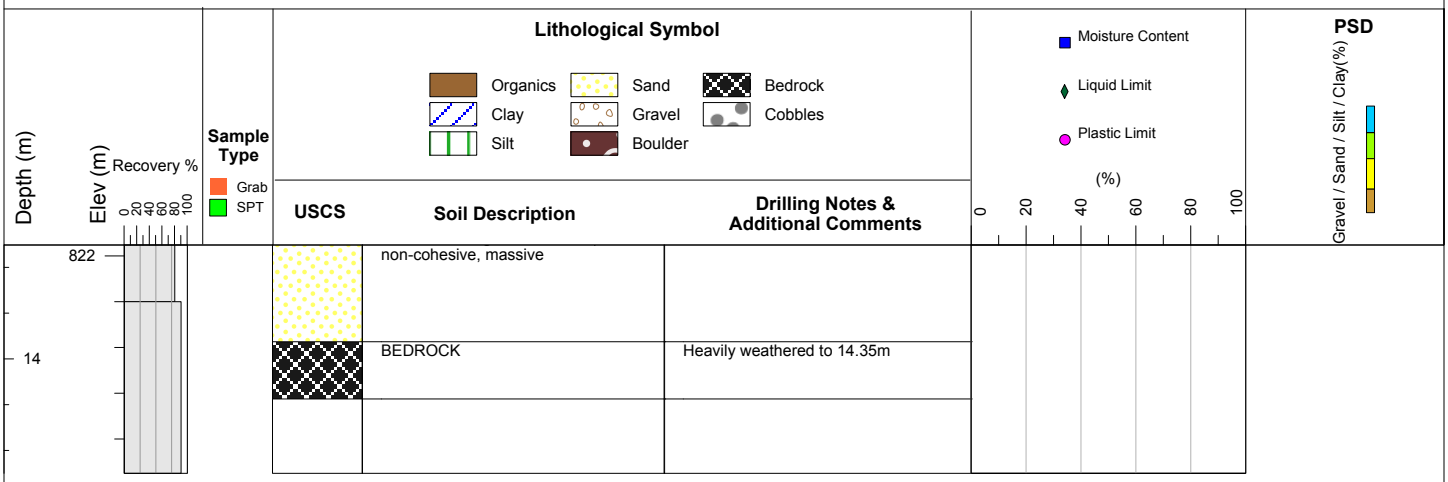
PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-970**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385125.3 E 6944272 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	835.1
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	820.75
TOTAL DEPTH (m):	14.35

DEPOSIT AREA: Area 2



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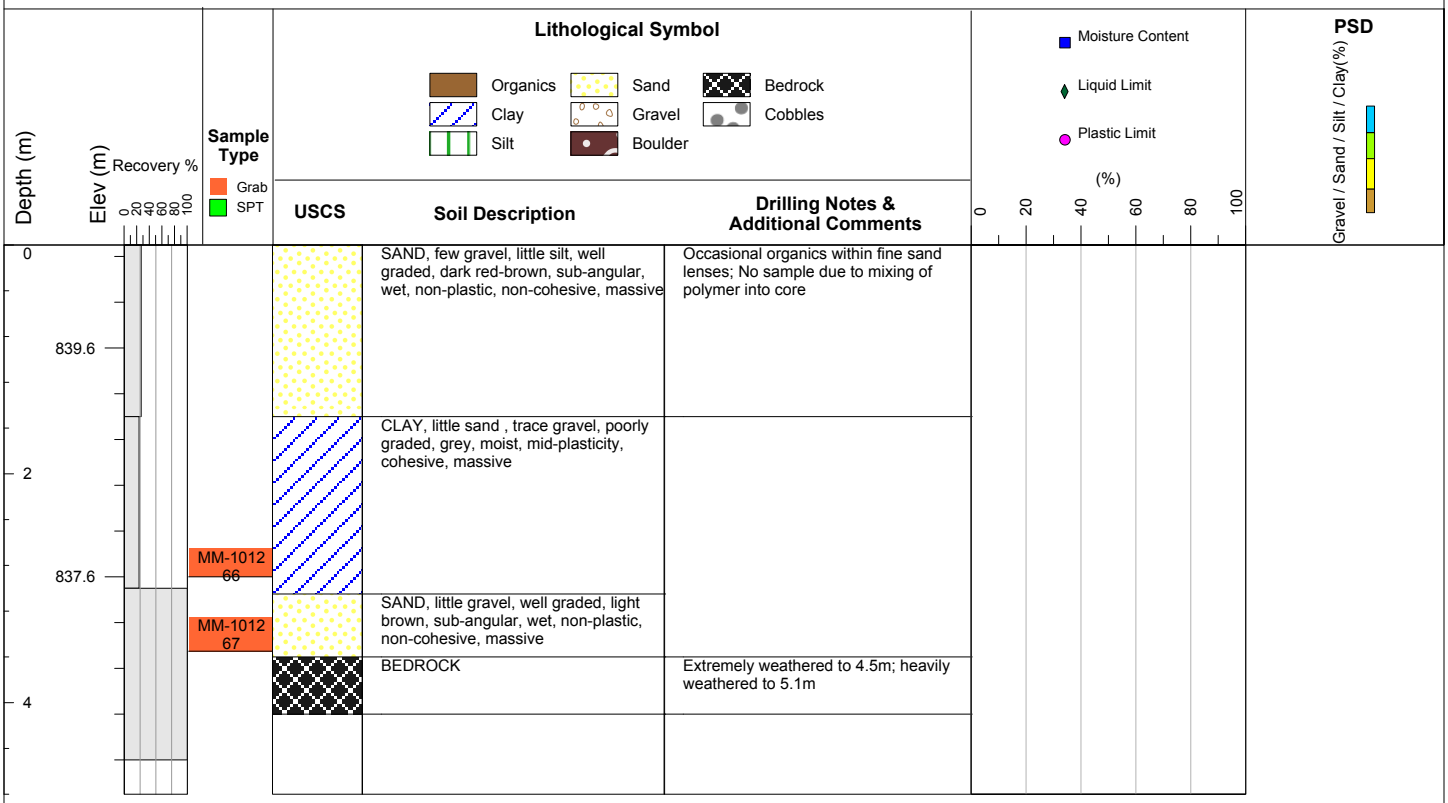
PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-971**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385075.1 E 6944288 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	840.5
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	836.4
TOTAL DEPTH (m):	4.1

DEPOSIT AREA: Area 2



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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-972**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385215.6 E 6944284 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	822.4
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	797.1
TOTAL DEPTH (m):	25.3

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%)

PSD Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery %	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
0	822				SAND , few gravel, some silt, well graded, Brownish Grey, sub-angular, moist, non-plastic, cohesive, massive	
2	820					
4	818		MM-1012 75			
6	816				SILT, some clay, few gravel, little sand , well graded, Grey, angular, moist, low-plasticity, cohesive, massive	
8	814		MM-1012 76		SILT, some clay, little gravel, some sand , well graded, Grey, sub-angular, moist, low-plasticity, cohesive, massive	
			MM-1012 77		SILT, some sand , trace gravel, well graded, Brown grey, sub-angular, wet, non-plastic, non-cohesive, massive	
10	812					
12			MM-1012 78			Chunk of cobble, and 3 sections of organic material 10 cm thick
	810				SAND, few silt, little clay, well graded, brown/grey, sub-rounded, moist, non-plastic, non-cohesive, massive	

11/32/49/8

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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-972**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385215.6 E 6944284 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	822.4
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	797.1
TOTAL DEPTH (m):	25.3

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%)

PSD Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery %	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
14	808		MM-101279			
16	806					
18	804		MM-101280		SILT, few sand , poorly graded, brown, wet, non-plastic, cohesive, massive	
20	802		MM-101281		SAND, little silt, poorly graded, brown, wet, non-plastic, non-cohesive, massive	Primarily fine sand
22	800		MM-101282		SAND, few silt, few gravel, well graded, brown, wet, non-plastic, cohesive, massive	Increased gravel content in bottom of this material unit
24	798		MM-101283		SAND, some gravel, well graded, light brown, wet, non-plastic, non-cohesive, massive	
26					BEDROCK	Heavily weathered to 28m

PSD: 0/4/95/1

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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-973**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385164.4 E 6944349 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	824
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	797
TOTAL DEPTH (m):	27

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%)

PSD Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery %	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
0	824		MM-101284		SILT, some clay, few gravel, poorly graded, grey/broun, sub-rounded, moist, mid-plasticity, cohesive, massive	Hole 14-SWC-973 has occasional ice lenses <2cm thick to 5m depth; approx. 5% excess ice; occasional cobbles; actual length of
2	822		MM-101285		CLAY, trace silt, trace gravel, poorly graded, Grey, sub-angular, moist, high-plasticity, cohesive, massive	Actual length of core 1.34m
4	820		MM-101286		Silt, some clay, some sand , little organics, well graded, grey/brown, moist, low-plasticity, cohesive, massive	Actual length of core 2.06m
6	818		MM-101287		SAND, some clay and silt, little gravel, well graded, brown/grey, sub-rounded, wet, non-plastic, non-cohesive, massive	Actual length of core 1.72m
8	816		MM-101288		SILT, some clay, few gravel, trace sand , well graded, grey/brown, sub-angular, moist, mid-plasticity, cohesive, blocky	Actual length same as run block
10	814					
12	812		MM-101289		GRAVEL, few sand , trace clay and silt, poorly graded, reddisg brown, sub-angular, moist, non-plastic, non-cohesive	

PSD: 0/13/63/24

PSD: 20/32/40/8

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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-973**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385164.4 E 6944349 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	824
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	797
TOTAL DEPTH (m):	27

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%) 0 20 40 60 80 100

PSD Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery % 0 20 40 60 80 100	Sample Type (Grab, SPT)	USCS	Soil Description	Drilling Notes & Additional Comments
14	810				GRAVEL, little sand , poorly graded, grey/black brown, angular, moist, non-plastic, non-cohesive	No sample
16	808				GRAVEL, some sand , trace silt, well graded, brown, angular, moist, non-plastic, non-cohesive, massive	One sample from box
18	806					
20	804					
22	802		MM-101290			
24	800		MM-101291		GRAVEL, few sand , trace silt, poorly graded, brown, angular, moist, non-plastic, non-cohesive, massive	
26	798					

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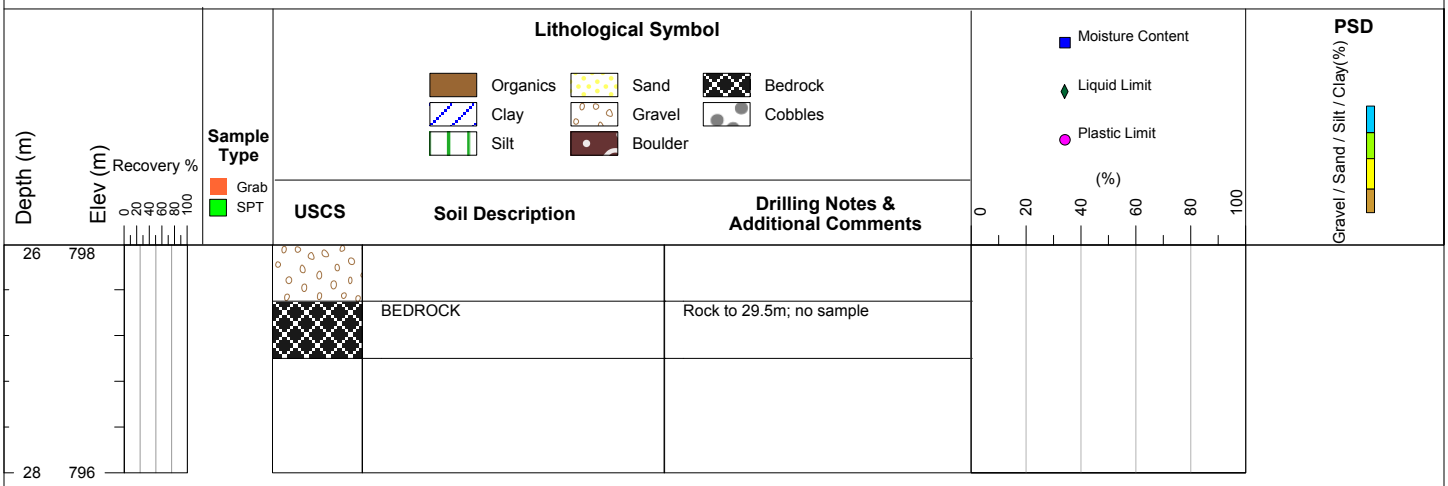
PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-973**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385164.4 E 6944349 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	824
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	797
TOTAL DEPTH (m):	27

DEPOSIT AREA: Area 2



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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-974**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385140 E 6944379 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	824.8
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	811.3
TOTAL DEPTH (m):	13.5

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%)

PSD: Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery %	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
0	824		MM-101292 (Grab)		SILT, little organics, little sand , poorly graded, grey, sub-rounded, wet, low-plasticity, cohesive, massive	Actual length of core 1.5m
2	822		MM-101293 (Grab)		CLAY, few silt, trace sand , poorly graded, grey, sub-rounded, wet, low-plasticity, cohesive, massive	
4	820		MM-101294 (Grab)		CLAY, little silt, trace sand , poorly graded, grey/brown, angular, moist, mid-plasticity, cohesive, massive	Actual length of core 2.5m
6	818		MM-101295 (Grab)		CLAY AND SILT, trace sand , little gravel, poorly graded, grey, sub-rounded, moist, mid-plasticity, cohesive, massive	Actual length of core 2.1m
8	816		MM-101296 (Grab)		BEDROCK, few gravel, little sand , poorly graded, red/brown, angular, moist, non-plastic, non-cohesive, massive	Actual length of core 4.5m
10	814					
12	812				BEDROCK, little gravel, trace sand , poorly graded, red/brown, angular, moist, non-plastic, non-cohesive, massive	Tagging error: 12.5 to 13.5. Actual length of core = 2.75m

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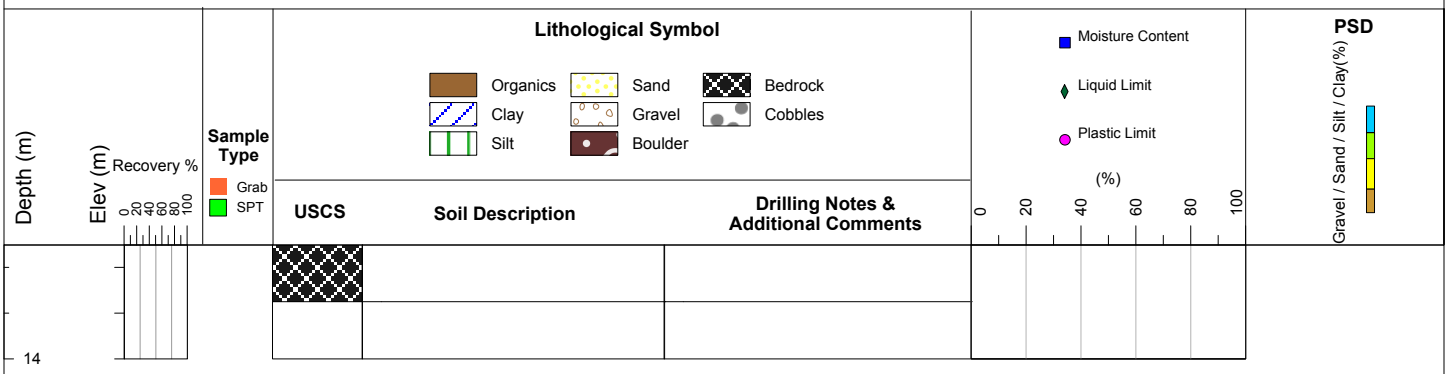
PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-974**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385140 E 6944379 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	824.8
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	811.3
TOTAL DEPTH (m):	13.5

DEPOSIT AREA: Area 2



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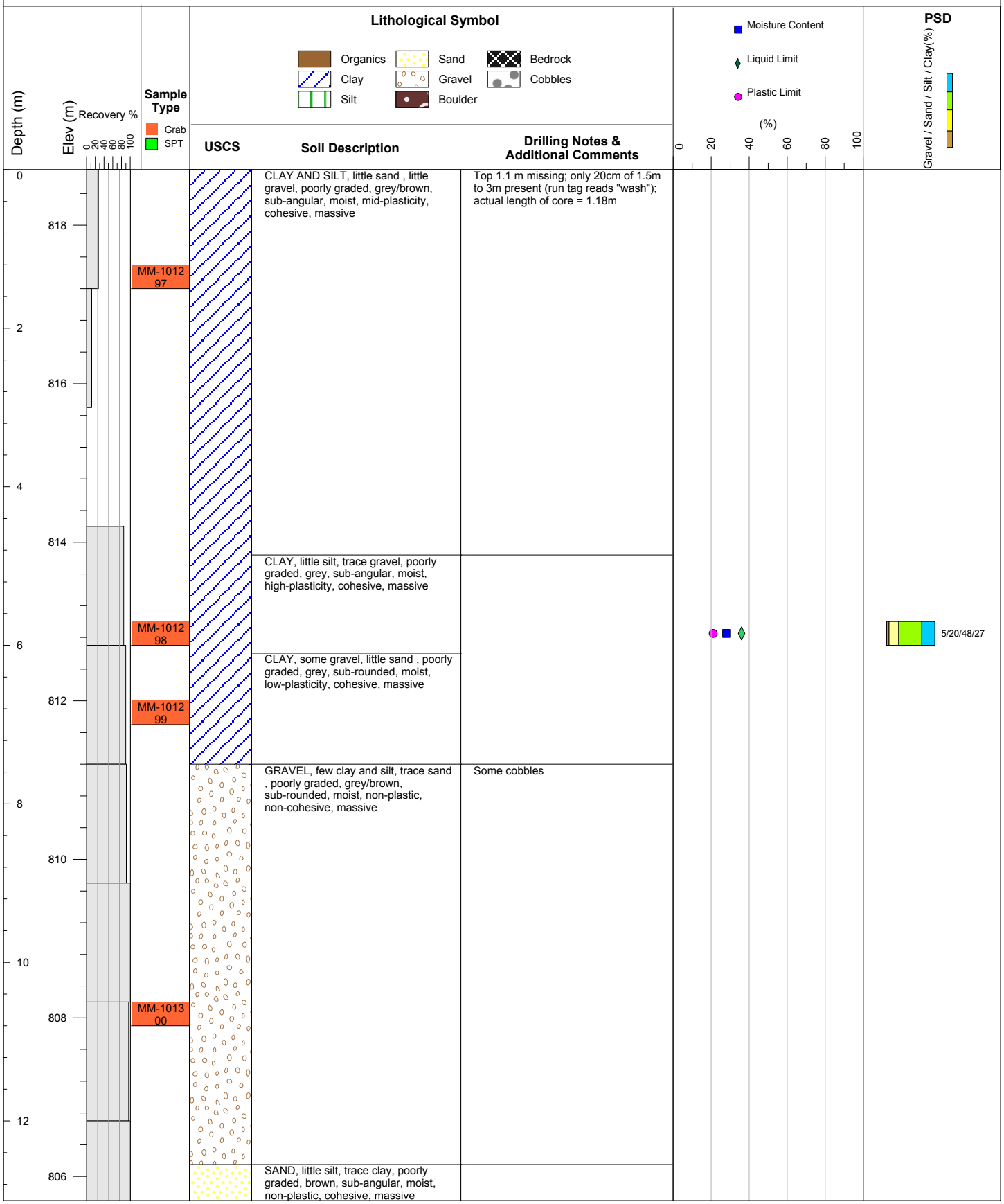
PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-975**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385156.1 E 6944428 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	818.7
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	796.3
TOTAL DEPTH (m):	22.4

DEPOSIT AREA: Area 2



USCS	Soil Description	Drilling Notes & Additional Comments
	CLAY AND SILT, little sand , little gravel, poorly graded, grey/brown, sub-angular, moist, mid-plasticity, cohesive, massive	Top 1.1 m missing; only 20cm of 1.5m to 3m present (run tag reads "wash"); actual length of core = 1.18m
	CLAY, little silt, trace gravel, poorly graded, grey, sub-angular, moist, high-plasticity, cohesive, massive	
	CLAY, some gravel, little sand , poorly graded, grey, sub-rounded, moist, low-plasticity, cohesive, massive	
	GRAVEL, few clay and silt, trace sand , poorly graded, grey/brown, sub-rounded, moist, non-plastic, non-cohesive, massive	Some cobbles
	SAND, little silt, trace clay, poorly graded, brown, sub-angular, moist, non-plastic, cohesive, massive	

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HOLE ID:	**14-SWC-975**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385156.1 E 6944428 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	818.7
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	796.3
TOTAL DEPTH (m):	22.4

PROJECT: Minto

CLIENT: Minto Exploration Ltd.

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit

PSD: Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery %	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
14	804		MM-1013 01			
			MM-1013 02			
16	802				GRAVEL, some sand , trace silt, poorly graded, brown, angular, moist, non-plastic, non-cohesive, massive	Sections of competent and weathered rock
18	800					
20	798		MM-1013 03			
22	796					
24	794					

(%) 0 20 40 60 80 100

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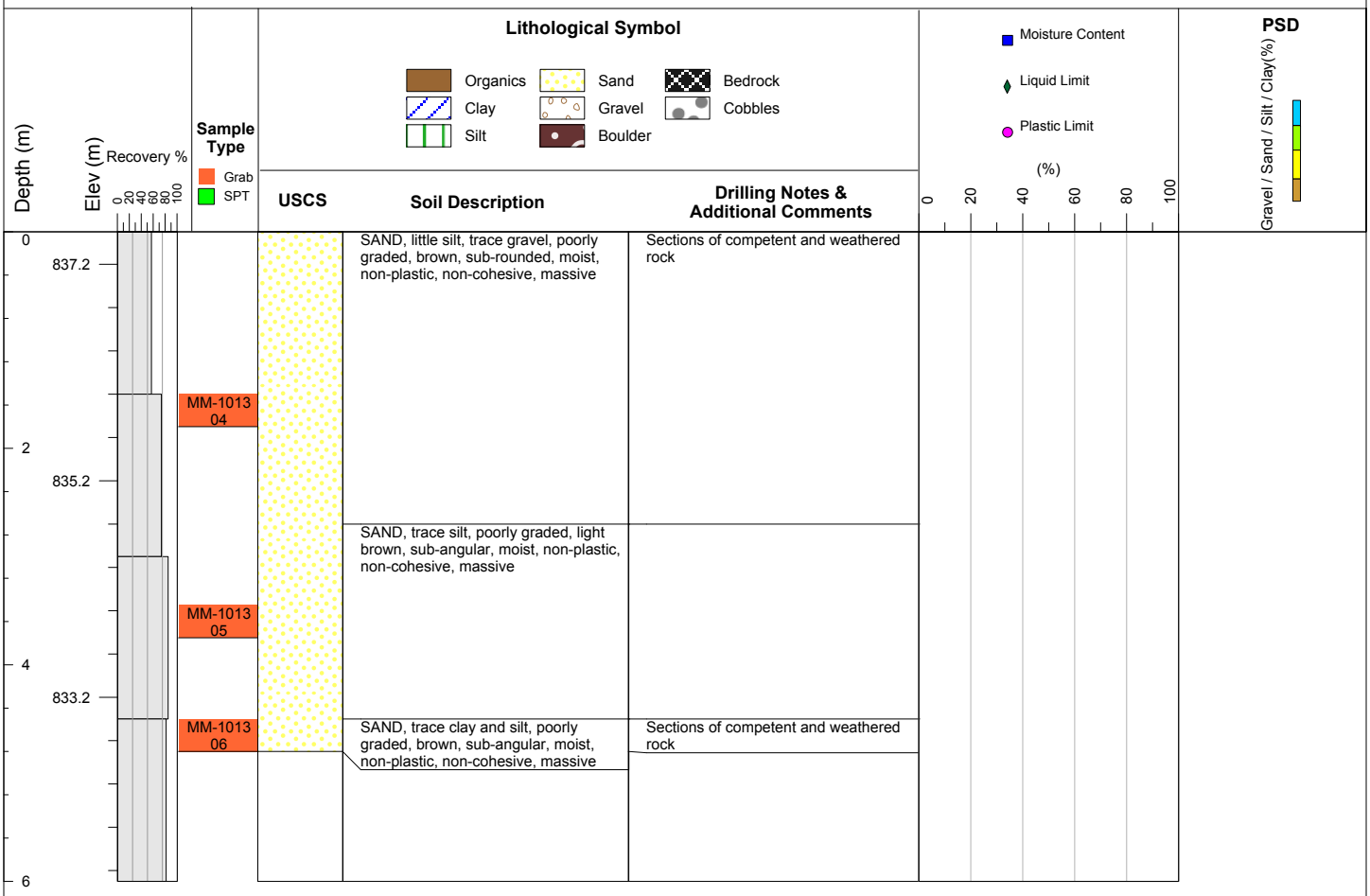
PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-976**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385075.3 E 6944328 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	837.5
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	832.77
TOTAL DEPTH (m):	4.8

DEPOSIT AREA: Area 2



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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-977**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385097.9 E 6944254 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	839.4
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	827.3
TOTAL DEPTH (m):	12.1

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%) 0 20 40 60 80 100

PSD Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery % (0 20 40 60 80 100)	Sample Type (Grab, SPT)	USCS	Soil Description	Drilling Notes & Additional Comments
0	839				SILT, some clay, few sand , trace gravel, poorly graded, grey brown, sub-angular, moist, mid-plasticity, cohesive, massive	Long interval of ablation till
2	837					
4	835		MM-101307			
6	833					
8	831		MM-101308			
10	829				SAND, few clay and silt, poorly graded, reddish grey brown, sub-rounded, moist, non-plastic, non-cohesive, massive	
12	827		MM-101309			

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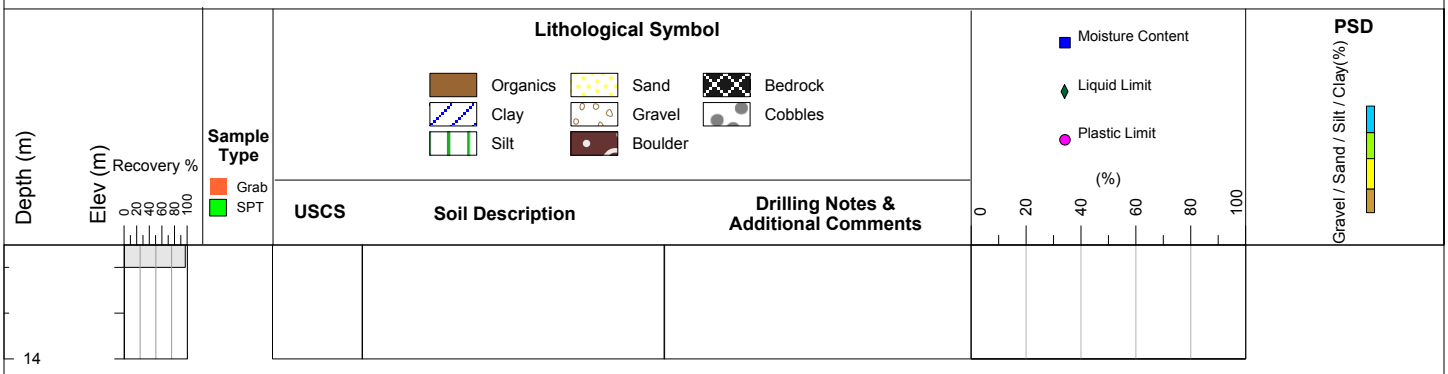
PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-977**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385097.9 E 6944254 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	839.4
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	827.3
TOTAL DEPTH (m):	12.1

DEPOSIT AREA: Area 2



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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-978**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385122.4 E 6944219 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	837.2
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	825.9
TOTAL DEPTH (m):	11.3

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%)

PSD Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery %	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
0	837					
2	835					
4	833					
6	831					
8	829		MM-101310		CLAY AND SILT, some sand , little gravel, poorly graded, brown matrix, sub-angular, moist, mid-plasticity, cohesive, massive	Diamictic ablational till, frequent granitic cobble, contact sharp in washout zone with underlying granodioritic regolith
10	827		MM-101311		CLAY AND SILT, little sand , trace gravel, poorly graded, brown matrix, sub-rounded, moist, mid-plasticity, cohesive, massive	
12	825					

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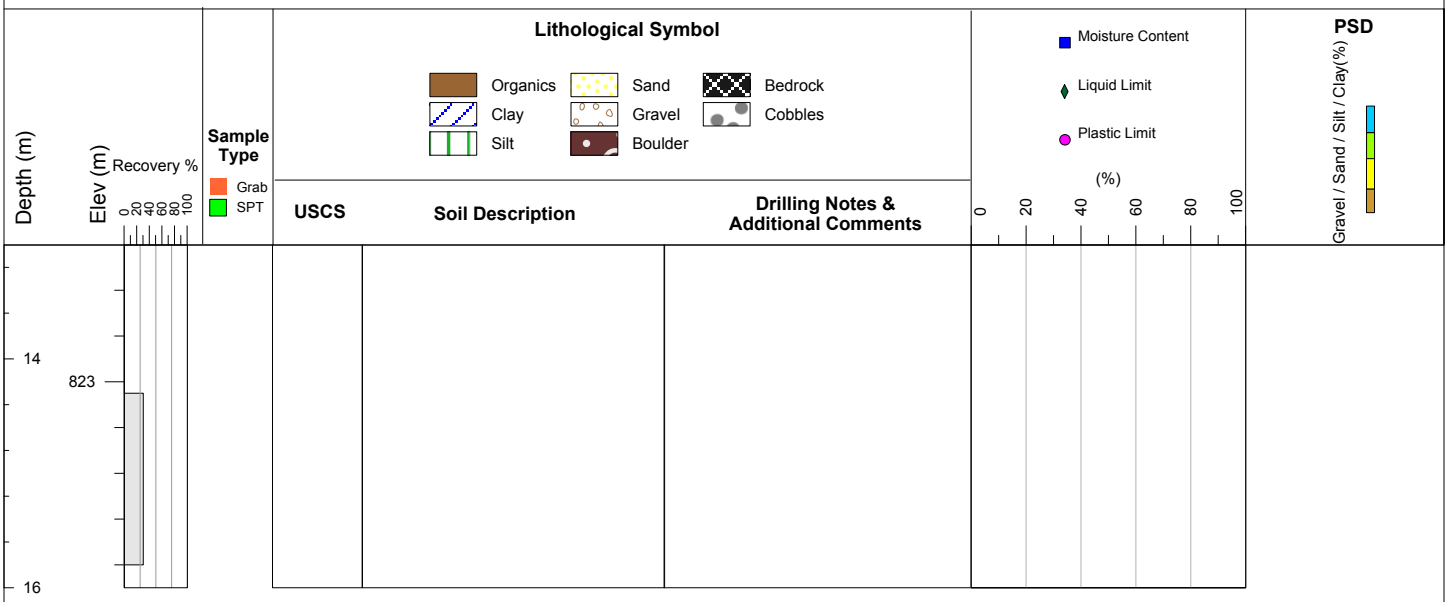
PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-978**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385122.4 E 6944219 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	837.2
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	825.9
TOTAL DEPTH (m):	11.3

DEPOSIT AREA: Area 2



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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-979**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385218.8 E 6944233 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	826.9
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	795.4
TOTAL DEPTH (m):	31.5

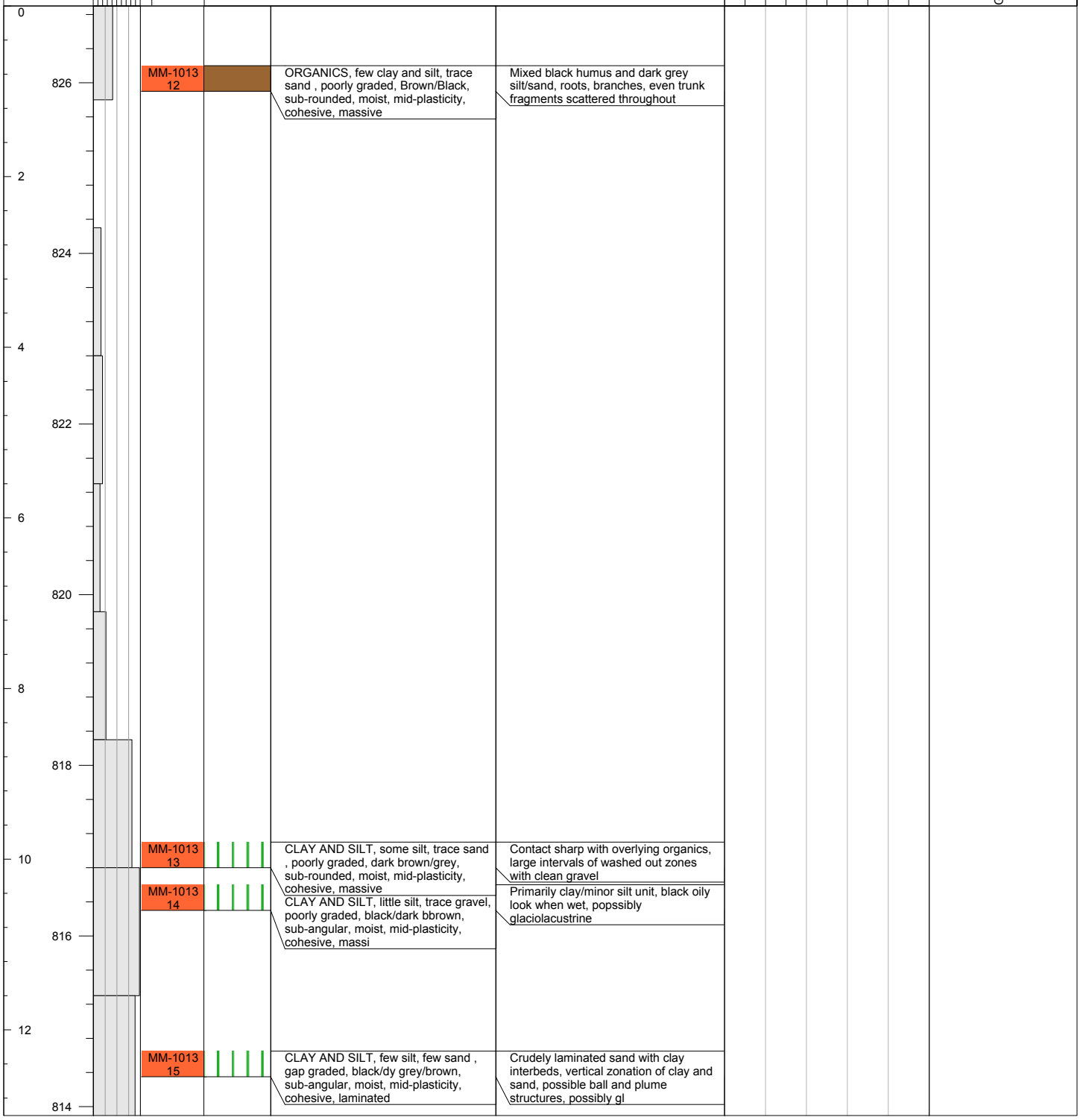
DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%)

PSD: Gravel / Sand / Silt / Clay(%)



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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-979**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385218.8 E 6944233 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	826.9
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	795.4
TOTAL DEPTH (m):	31.5

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%) 0 20 40 60 80 100

PSD Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery % (0 20 40 60 80 100)	Sample Type (Grab, SPT)	USCS	Soil Description	Drilling Notes & Additional Comments
14			MM-1013 16		SILT, few sand , few gravel, poorly graded, grey brown, sub-angular, moist, non-plastic, non-cohesive, massive	Ablation till - granular to cobble lithic and volcanic clasts in a silty sandy matrix
16	812					
18	810					
20	808					
22	806		MM-1013 17		CLAY AND SILT, some clay, trace sand , poorly graded, brown, sub-angular, wet, non-plastic, non-cohesive, massive	Massive nondescript clay/silt unit, localized saturation; possibly glaciolacustrine
24	804					
26	802		MM-1013 18		SILT, few clay, trace sand , poorly graded, brown, rounded, moist,	Massive nondescript silt unit; possibly glaciolacustrine

PSD: 0/21/77/2

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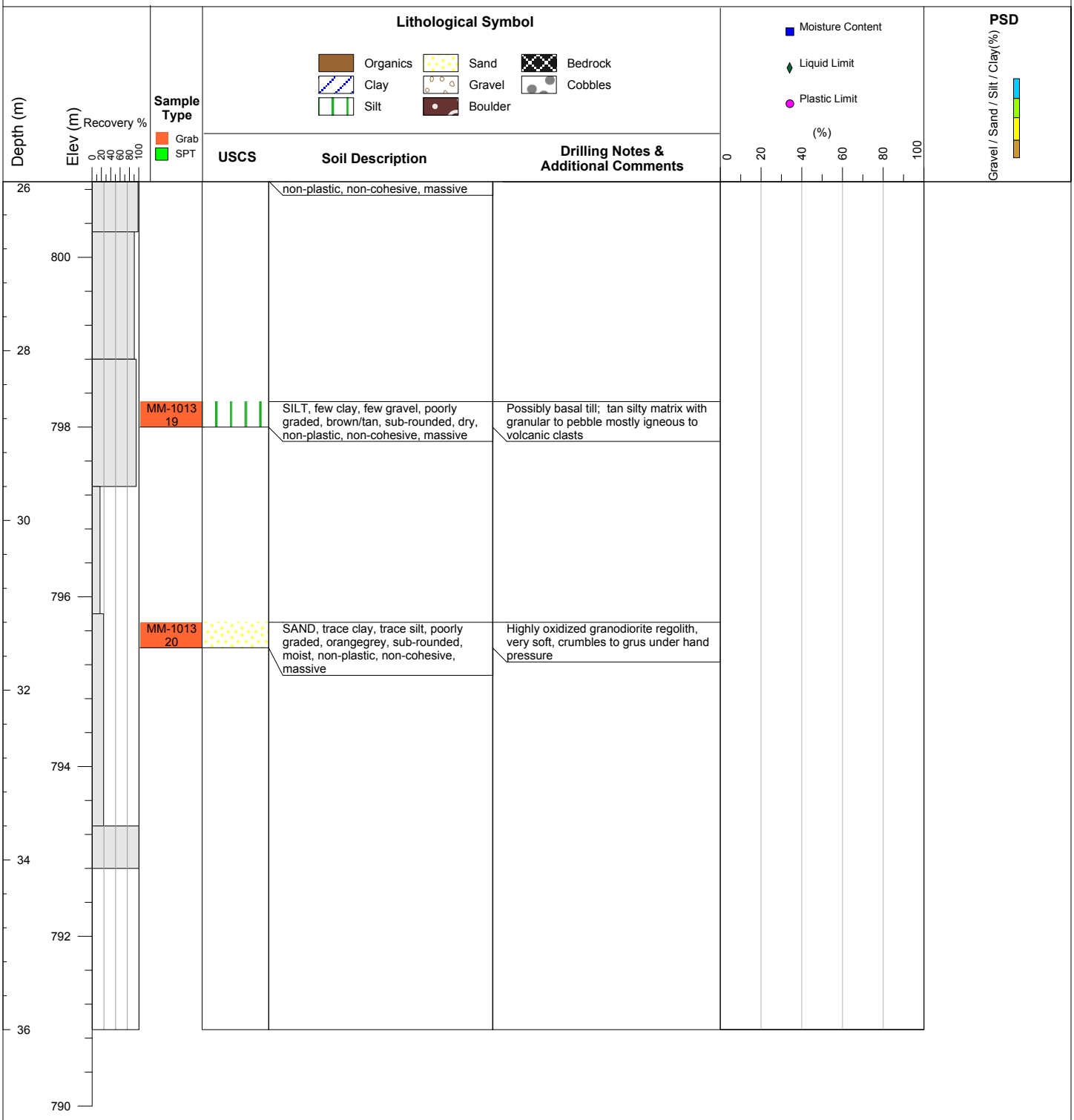
PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-979**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385218.8 E 6944233 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	826.9
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	795.4
TOTAL DEPTH (m):	31.5

DEPOSIT AREA: Area 2



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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-980**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385194.5 E 6944261 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	826.5
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	804.9
TOTAL DEPTH (m):	21.6

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%)

PSD Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery %	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
0	826					
2	824					
4	822		MM-1013 21		SILT, few sand , few gravel, poorly graded, brown, sub-rounded, moist, non-plastic, cohesive, massive	Diamictic till; silty matrix
6	820		MM-1013 22		CLAY, few sand , little gravel, poorly graded, grey/brown, sub-rounded, moist, mid-plasticity, cohesive, massive	Massive clay with lenses of sand; bedding non apparent
8	818					
10	816		MM-1013 23		CLAY, some silt, little gravel, poorly graded, grey/brown, sub-angular, moist, mid-plasticity, cohesive, massive	Clay with discontinous sand lenses; bedding non-apparent
12	814					

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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-980**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385194.5 E 6944261 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	826.5
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	804.9
TOTAL DEPTH (m):	21.6

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%) 0 20 40 60 80 100

PSD Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery % (0 20 40 60 80 100)	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
14	812		MM-1013 24 (Grab)		CLAY AND SILT, few sand , trace gravel, poorly graded, brown/grey, sub-rounded, moist, mid-plasticity, cohesive, massive	Same as above, clay with silty, sandy discontinous lenses, infrequent granule to cobble volcanic, lithic and igneous clasts
16	810					
18	808		MM-1013 25 (SPT)		SILT, few sand , trace gravel, poorly graded, brown, sub-angular, moist, non-plastic, non-cohesive, massive	Non-descript massive silty unit, few clasts; possibly glaciolacustrine
20	806		MM-1013 26 (SPT)		CLAY, little silt, trace sand , poorly graded, brown, sub-angular, moist, mid-plasticity, cohesive, massive	Massive clay; possibly glaciolacustrine
22	804		MM-1013 27 (SPT)		SILT, some sand , little gravel, poorly graded, brown/orange, sub-angular, moist, non-plastic, non-cohesive, massive	Sandy ablation till with fragments of granodirite - malachite; angular sharp contact with underlying bedrock (granodiorite) at
24	802					

PSD: 6/40/52/2

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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-981**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385148 E 6944188 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	837.838
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	807.89
TOTAL DEPTH (m):	29.95

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Sand, Bedrock, Clay, Gravel, Cobbles, Silt, Boulder

Sample Type: Grab, SPT

Recovery % (0, 20, 40, 60, 80, 100)

Moisture Content, Liquid Limit, Plastic Limit (%) (0, 20, 40, 60, 80, 100)

PSD: Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	USCS	Soil Description	Drilling Notes & Additional Comments
0			SAND, few silt, trace gravel, poorly graded, dark brown, sub-rounded, moist, non-plastic, non-cohesive, massive	Organic matter (tree root) for initial 10 cm
2	836		SAND, few gravel, poorly graded, brown, sub-rounded, moist, non-plastic, non-cohesive, massive	No recovery (loss of matrix sand) from 5.4-6.9m
4	834			
6	832			
8	830		CLAY AND SILT, few sand , little gravel, poorly graded, grey brown, sub-rounded, moist, mid-plasticity, cohesive, massive	
10	828		SAND, some gravel, trace silt, poorly graded, brown, sub-rounded, moist, non-plastic, non-cohesive, massive	
12	826		CLAY AND SILT, few gravel, little sand , poorly graded, dark grey, sub-rounded, moist, mid-plasticity, cohesive, massive	Minor organic matter including twigs

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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-981**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385148 E 6944188 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	837.838
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	807.89
TOTAL DEPTH (m):	29.95

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%)

PSD: Gravel / Sand / Silt / Clay(%)

Sample Type: Grab, SPT

Recovery % (0, 20, 40, 60, 80, 100)

Depth (m)	Elev (m)	USCS	Soil Description	Drilling Notes & Additional Comments
14	824			
16	822		SAND, few silt, little gravel, poorly graded, brown, sub-rounded, moist, non-plastic, non-cohesive, massive	
			CLAY AND SILT, little sand , little gravel, poorly graded, dark grey, sub-rounded, moist, mid-plasticity, cohesive, massive	
18	820		SAND, few clay and silt, little gravel, poorly graded, light brown, sub-rounded, moist, non-plastic, non-cohesive, massive	
20	818		CLAY AND SILT, some gravel, little sand , poorly graded, grey, sub-rounded, moist, mid-plasticity, cohesive, massive	
			SAND, little gravel, poorly graded, brown, sub-rounded, moist, non-plastic, non-cohesive, massive	
22	816			
24	814		CLAY AND SILT, few sand , trace gravel, poorly graded, grey brown, sub-angular, moist, low-plasticity, cohesive, massive	Very coarse gravel with clasts up to 25m
			SAND, some gravel, trace silt, poorly graded, dark brown, sub-rounded, moist, non-plastic, non-cohesive, massive	
26	812		SAND, few gravel, little clay, poorly graded, maroon, sub-angular, moist, non-plastic, cohesive, massive	

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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**14-SWC-981**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM
BORING DATE:	Oct/Nov 2014

COORDINATES:	385148 E 6944188 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	837.838
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	807.89
TOTAL DEPTH (m):	29.95

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%) 0 20 40 60 80 100

PSD Gravel / Sand / Silt / Clay(%)

Depth (m) | Elev (m) | Recovery % 0 20 40 60 80 100 | Sample Type: Grab, SPT

Depth (m)	Elev (m)	USCS	Soil Description	Drilling Notes & Additional Comments
26				
			SAND, some gravel, little clay, poorly graded, grey, sub-angular, moist, non-plastic, cohesive, massive	Local highly oxidized limoinitic fragments; some with malachite
	810			
28			GRAVEL, little sand , maroon, sub-rounded, moist, non-plastic, non-cohesive, massive	Fines almost completely washed away; only cobbles remaining
			SAND, little gravel, poorly graded, maroon, angular, moist, non-plastic, non-cohesive, massive	Probably strongly altered rock but behaving as a soil
	808			
30				
	806			

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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**15-SWC-995**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM/DM
BORING DATE:	5-Feb-15 To 9-Feb-15

COORDINATES:	385096.08 E 6944251.64 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	839.41
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	826.66
TOTAL DEPTH (m):	13.2

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%) 0 20 40 60 80 100

PSD Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery % (0 20 40 60 80 100)	Sample Type (Grab, SPT)	USCS	Soil Description	Drilling Notes & Additional Comments
0	839				SAND, few gravel, some silt, well graded, brown, sub-angular, wet, low plasticity, cohesive, massive	Occasional cobbles
2	838					
	837		58552			
	836					
4					CLAY, few silt, some sand, poorly graded, brown, , wet, med plasticity, cohesive, massive	
	835		58553			
	834				SILT, few sand, few gravel, trace clay, well graded, brown, sub-angular, wet, low plasticity, cohesive, massive	Occasional cobbles
6	833				SILT, some sand, little gravel, little clay, well graded, brown, sub-rounded, wet, non-plastic, non-cohesive, massive	Cohesive fines; occasional cobbles
	832					
8	831		58554			
	830				SILT, some sand , little gravel, little clay, well graded, brown, sub-rounded, wet, non-plastic, non-cohesive, massive	
10	829		58555		SAND, some gravel, well graded, reddish-brown, sub-angular, wet, non-plastic, non-cohesive, massive	Extremely weathered diorite. Bedrock at 13.2m
	828					
12	827					

PSD: 12/36/43/9

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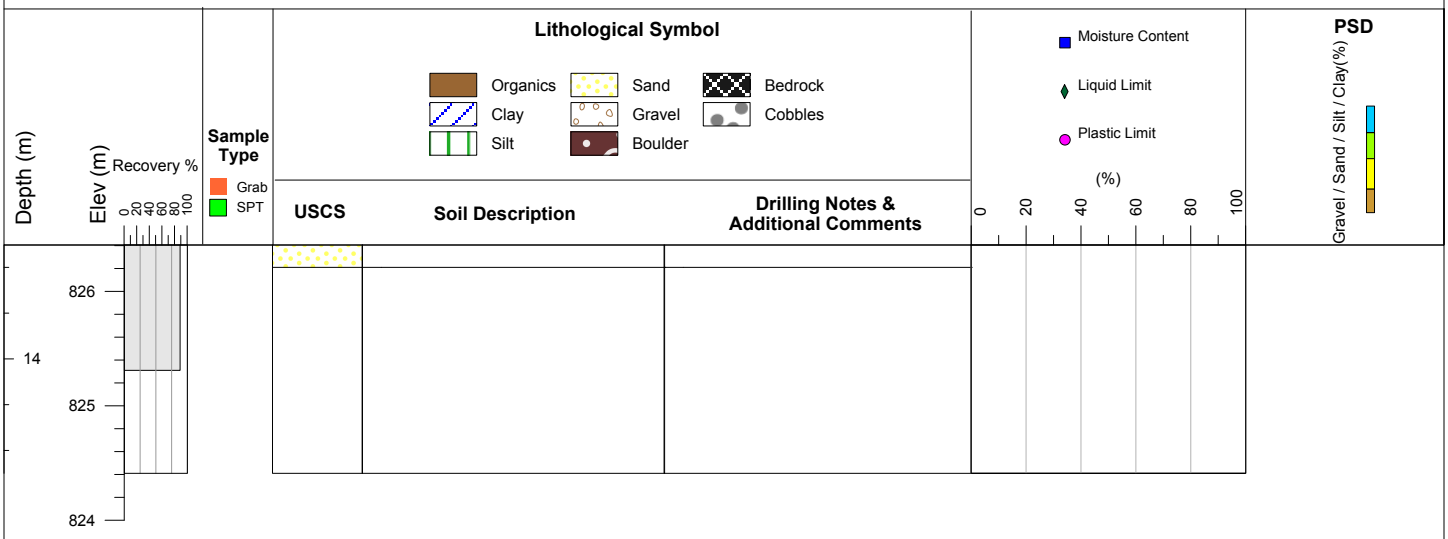
PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**15-SWC-995**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM/DM
BORING DATE:	5-Feb-15 To 9-Feb-15

COORDINATES:	385096.08 E 6944251.64 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	839.41
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	826.66
TOTAL DEPTH (m):	13.2

DEPOSIT AREA: Area 2



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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**15-SWC-996**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM/DM
BORING DATE:	9-Feb-15 To 11-Feb-15

COORDINATES:	385161 E 6944314.88 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	827.87
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	803.45
TOTAL DEPTH (m):	28.2

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%) 0 20 40 60 80 100

PSD Gravel / Sand / Silt / Clay(%)

Recovery % 0 20 40 60 80 100

Depth (m)	Elev (m)	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
0				SILT, few sand, few gravel, little organics, poorly graded, dark brown, sub-rounded, moist, low plasticity, cohesive, massive	
		58556		SAND, some silt, few gravel, trace boulder, poorly graded, brown, sub-rounded, moist, low plasticity, cohesive, massive	Occasional cobbles up to 13cm
2	826				
4	824			CLAY, few silt, well graded, dark brown-grey, rounded, moist, high plasticity, cohesive, massive	Unable to determine exact contact as next run was not fully recovered
				SILT, few gravel, some clay, poorly graded, mottled grey-brown, sub-rounded, , low plasticity, non-cohesive, massive	Unkown parameters due to low recovery
6	822	58557		SILT, some sand, little gravel, well graded, light brown, sub-angular, moist, med plasticity, cohesive, bedded >3cm	Bedded at 45-55 deg tca
8	820			SILT, some sand, few gravel, little boulder, poorly graded, brown, sub-angular, moist, high plasticity, cohesive, blocky	Poorly sorted, local cobbles up to 11m; locally bedded but still poorly sorted
10	818	58558		SILT, some sand, little gravel, well graded, dark/light brown, sub-rounded, moist, med plasticity, cohesive, thinly bedded >0.5	
12	816				No recovery; Lost Core
				SAND, few gravel, few silt, poorly graded, brown-grey, rounded, moist,	

PSD: 16/35/43/6

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capstone MINTO MINE

PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**15-SWC-996**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM/DM
BORING DATE:	9-Feb-15 To 11-Feb-15

COORDINATES:	385161 E 6944314.88 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	827.87
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	803.45
TOTAL DEPTH (m):	28.2

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%) 0 20 40 60 80 100

PSD Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery % 0 20 40 60 80 100	Sample Type (Grab, SPT)	USCS	Soil Description	Drilling Notes & Additional Comments
					low plasticity, cohesive, blocky	
14	814				SAND, few silt, few gravel, well graded, brown, sub-angular, moist, low plasticity, cohesive, bedded >3cm	Mod-well bonded; bedding at 50 deg tca. 15.20 - 15.54m; soft mud/clay/sand mix, appears to be artificial/drill induced.
			58559			
					SILT, few sand, little gravel, well graded, brown-grey, sub-rounded, wet, low plasticity, cohesive, massive	Some wash out from 20.2-21.7 likely caused due to reeming since bit started off bottom on this run
16	812				SAND, some silt, few gravel, poorly graded, brown, sub-rounded, wet, med plasticity, cohesive, massive	
					SILT, some sand, little gravel, poorly graded, brown, sub-rounded, moist, low plasticity, cohesive, massive	Significant colour change at end of unit
18	810					
			58560			
20	808					
					SAND, few gravel, poorly graded, reddish brown, sub-angular, moist, low plasticity, cohesive, massive	Unit is composed of highly weathered bedrock material; brecciated appearance
22	806					
			58561			
					BEDROCK	
24	804					
26	802					

PSD: 11/30/51/8

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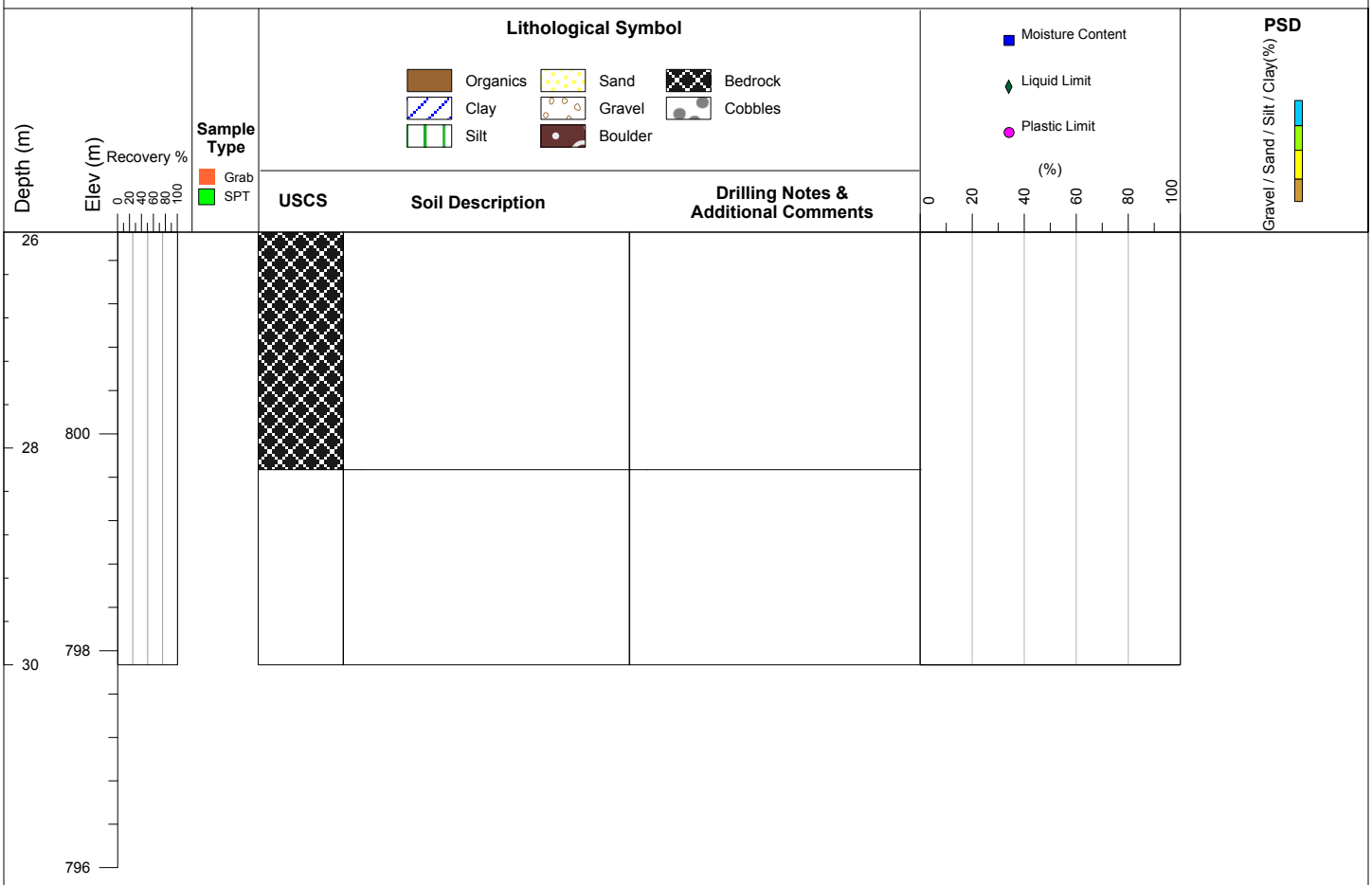
PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**15-SWC-996**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM/DM
BORING DATE:	9-Feb-15 To 11-Feb-15

COORDINATES:	385161 E 6944314.88 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	827.87
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	803.45
TOTAL DEPTH (m):	28.2

DEPOSIT AREA: Area 2



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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**15-SWC-997**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM/DM
BORING DATE:	12-Feb-15 To 14-Feb-15

COORDINATES:	385140.87 E 6944123.97 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	841.65
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	789.65
TOTAL DEPTH (m):	52.8

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%)

PSD: Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery %	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
0					GRAVEL, some sand, little silt, well graded, grey-brown, sub-rounded, wet, non-plastic, non-cohesive, massive	Occasional cobble; spoor recovery in this unit; evidence of cobble and gravels gettig stuck and spun in the bit which likely di
			58562 (Grab)			
	840					
2						
					GRAVEL, some sand, little silt, well graded, grey-brown, sub-rounded, wet, non-plastic, non-cohesive, massive	Assumed to be the same as recorded on the first hole; nothing suggests otherwise; no salt brine was used for the first 3.8m to
4						Poor recovery; indicates same a soil above; core barrel lot down hole requiring re-drill hole.
	836				GRAVEL, some sand, little silt, well graded, grey-brown, sub-rounded, wet, non-plastic, non-cohesive, massive	Some slush within split tube suggests brine remains subfreezing during drilling process
6					COBBLES, few gravel, few sand, few silt, sub-rounded, , non-plastic, non-cohesive,	Low recovery with mostly cobbles and gravel recovered with a 7cm section of sand/silt.
					non-plastic, non-cohesive,	No recovery
8						
					COBBLES, few gravel, few sand, few silt, sub-rounded, , non-plastic, non-cohesive,	Low recovery with mostly cobbles and gravel recovered with a 7cm section of sand/silt.
	832					
10			160025 (Grab)		SAND, few gravel, few silt, trace clay, poorly graded, light brown, sub-rounded, moist, low plasticity, cohesive, bedded >3cm	11.7-11.8m: 10cm silty band (with minor clay) lower cohesion, possibly larger (15cm lost recvery). Bedding locally apparent at
12			58563 (Grab)			

PSD: 30/42/24/4

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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**15-SWC-997**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM/DM
BORING DATE:	12-Feb-15 To 14-Feb-15

COORDINATES:	385140.87 E 6944123.97 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	841.65
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	789.65
TOTAL DEPTH (m):	52.8

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit

PSD: Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery %	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
14	828		58564		SILT, some sand, few gravel, trace boulder, poorly graded, light brown, sub-rounded, moist, low plasticity, cohesive, massive	Similar to previous unit but no bedding aparent.
16			58565			
			58566		CLAY, some silt, few sand, trace gravel, well graded, dark grey brown, sub-angular, moist, high plasticity, cohesive, thinly be	mm to cm scale clay rich beds, local sandier beds. Bedding at 68 deg tca.
18	824				SAND, some silt, little gravel, trace boulder, poorly graded, light grey brown, sub-angular, moist, low plasticity, cohesive, m	
20			58567		SAND, some silt, little organics, little clay, well graded, dark grey, sub-angular, moist, med plasticity, cohesive, thinly bed	18.9-19.13m: wood. Locally decomposed but mostly well preserved; immediately below this is a 6cm sand/gravel bed containing 15%
22	820				CLAY, some silt, few sand, trace organics, poorly graded, dark grey, sub-angular, moist, high plasticity, cohesive, massive	
					SAND, few silt, little gravel, well graded, grey, sub-angular, moist, non-plastic, cohesive, thinly bedded >0.5cm	Granite boulder
					BOULDER, pink-grey, sub-angular, dry, non-plastic, cohesive, blocky	
					CLAY, few silt, little sand, well graded, grey, sub-angular, moist, high plasticity, cohesive, massive	Trace organics present
24			58568		GRAVEL, few sand, few silt, little clay, poorly graded, grey, sub-angular, moist, non-plastic, cohesive, thinly bedded >0.5cm	
26	816				SAND, some silt, little gravel, trace boulder, well graded, grey-brown, sub-angular, moist, non-plastic, cohesive, thinly bedde	One 20cm boulder

PSD: 44/29/21/6

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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**15-SWC-997**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM/DM
BORING DATE:	12-Feb-15 To 14-Feb-15

COORDINATES:	385140.87 E 6944123.97 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	841.65
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	789.65
TOTAL DEPTH (m):	52.8

DEPOSIT AREA: Area 2

Lithological Symbol: Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%)

PSD: Gravel / Sand / Silt / Clay(%)

Depth (m)	Elev (m)	Recovery %	Sample Type	USCS	Soil Description	Drilling Notes & Additional Comments
26						
28			58569		SILT, some sand, little clay, trace organics, well graded, light-med brown, sub-angular, moist, med plasticity, cohesive, thinl	
					SAND, few gravel, little silt, poorly graded, mottled brown, sub-rounded, moist, non-plastic, cohesive, massive	Locally non-cohesive
	812					
30			58570		SAND, some silt, little gravel, well graded, med brown, sub-rounded, moist, non-plastic, cohesive, bedded >3cm	Very well graded; (coarse sand beds and fine silt beds) 75 deg tca bedding
32						
	808				SAND, few gravel, few silt, well graded, mottled brown, sub-rounded, moist, non-plastic, cohesive, bedded >3cm	
34			58571			
					SILT, few sand, little gravel, trace boulder, well graded, med brown, sub-rounded, moist, non-plastic, cohesive, bedded >3cm	
36						
	804					
38					SAND, few gravel, little silt, little boulder, poorly graded, mottled brown, sub-angular, moist, non-plastic, cohesive, blocky	Occasional cobble/boulder
					SAND, few gravel, little boulder, little	

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PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**15-SWC-997**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM/DM
BORING DATE:	12-Feb-15 To 14-Feb-15

COORDINATES:	385140.87 E 6944123.97 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	841.65
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	789.65
TOTAL DEPTH (m):	52.8

DEPOSIT AREA: Area 2

Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

Moisture Content, Liquid Limit, Plastic Limit (%) 0 20 40 60 80 100

PSD Gravel / Sand / Silt / Clay(%)

Depth (m) | Elev (m) | Recovery % 0 20 40 60 80 100 | Sample Type: Grab, SPT

Depth (m)	Elev (m)	Sample	USCS	Soil Description	Drilling Notes & Additional Comments
40				silt, poorly graded, mottled brown, sub-angular, moist, non-plastic, non-cohesive, blo	
				SILT, few clay, some sand, few gravel, poorly graded, grey, sub-angular, moist, med plasticity, cohesive, massive	
	800	58572			
42					
44					
	796				
46		58573		SILT, few clay, few gravel, few sand, poorly graded, grey, sub-rounded, moist, med plasticity, cohesive, massive	Unknown lower contact due to poor recovery, assumed to end early since this sort of soil has historically had better recovery t
		58571		GRAVEL, some sand, little silt, well graded, grey-brown, sub-angular, moist, non-plastic, non-cohesive, massive	Poor recovery throughout this unit
48					
	792				
50				SAND, little silt, few gravel, well graded, red-brown, sub-rounded, moist, non-plastic, cohesive, massive	
52					

PSD: 15/34/29/22

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capstone MINTO MINE

PROJECT: Minto

CLIENT: Minto Exploration Ltd.

HOLE ID:	**15-SWC-997**
LOCATION:	Yukon
PROJECT NO:	1CM002.037
DRILLING CONTRACTOR:	Driftwood
DRILLING TYPE & CORE DIA:	HQ Diamond
LOGGED BY:	MJM/DM
BORING DATE:	12-Feb-15 To 14-Feb-15

COORDINATES:	385140.87 E 6944123.97 N
DATUM:	UTM Zone 8
GROUND ELEV (m):	841.65
AZIMUTH:	0
COLLAR DIP:	-90
EOH ELEV. (m):	789.65
TOTAL DEPTH (m):	52.8

DEPOSIT AREA: Area 2

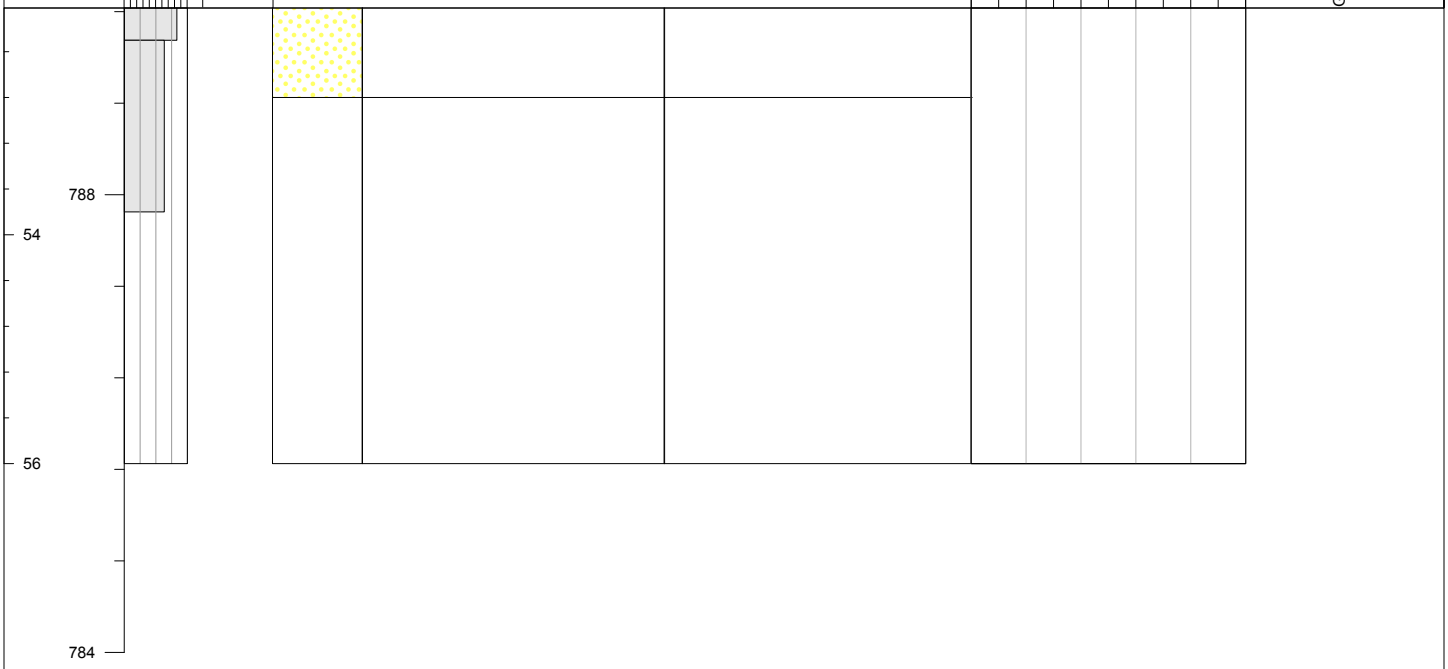
Lithological Symbol

Organics, Clay, Silt, Sand, Gravel, Boulder, Bedrock, Cobbles

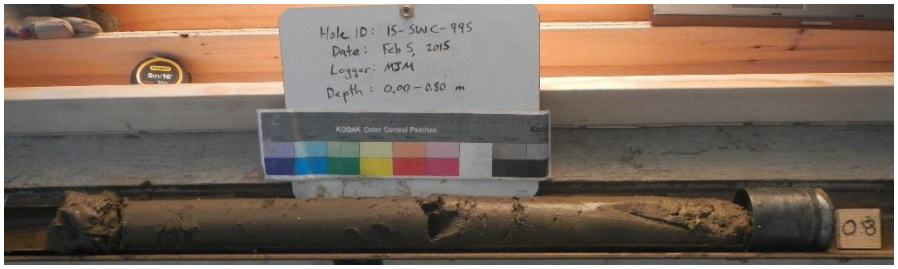
Moisture Content, Liquid Limit, Plastic Limit (%)

PSD Gravel / Sand / Silt / Clay(%)

Depth (m) | Elev (m) | Recovery % (0 20 40 60 80 100) | **Sample Type** Grab, SPT | **USCS** | **Soil Description** | **Drilling Notes & Additional Comments**



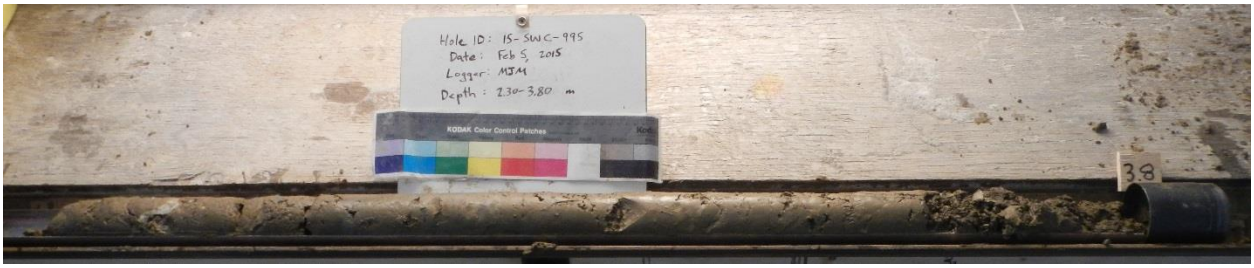
Attachment 2: Core Photos



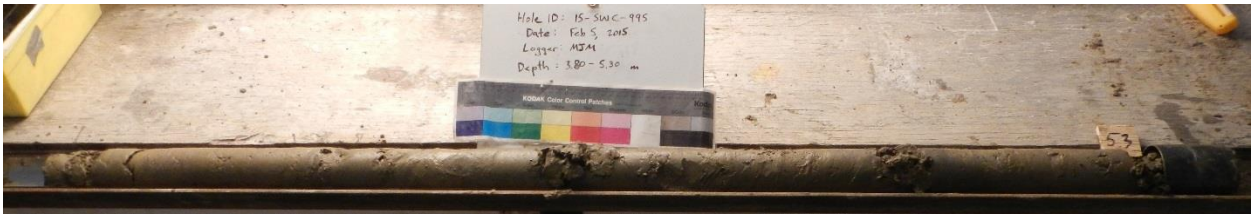
15-SWC-995: 0 m - 0.8 m



15-SWC-995: 0.8 m - 2.3 m



15-SWC-995: 2.3 m - 3.8 m



15-SWC-995: 3.8 m - 5.3 m



15-SWC-995: 5.3 m - 6.8 m



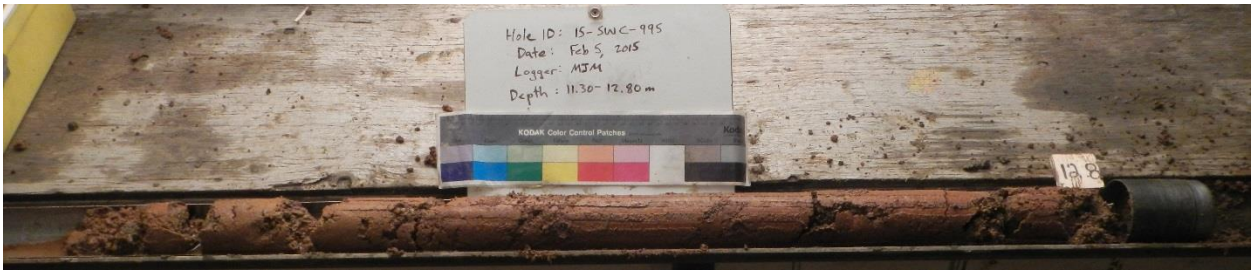
15-SWC-995: 6.8 m - 8.3 m



15-SWC-995: 8.3 m - 9.8 m



15-SWC-995: 9.8 m - 11.3 m



15-SWC-995: 11.3 m - 12.8 m



15-SWC-995: 12.8 m - 14.1 m



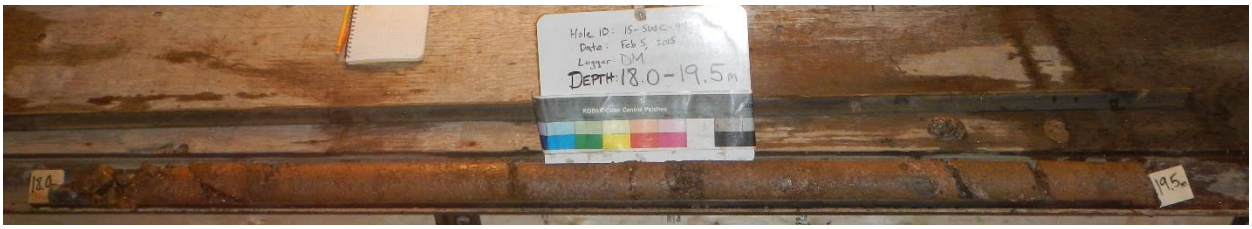
15-SWC-995: 14.1 m - 15.6 m



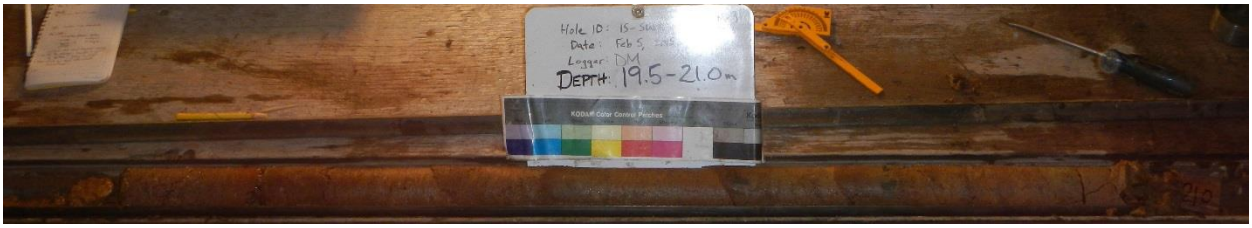
15-SWC-995: 15.6 m - 17.1 m



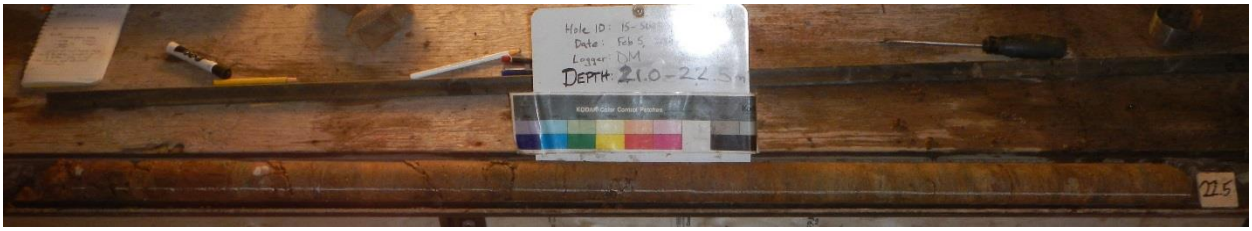
15-SWC-995: 17.1 m - 18 m



15-SWC-995: 18 m - 19.5 m



15-SWC-995: 19.5 m - 21 m



15-SWC-995: 21 m - 22.5 m



15-SWC-995: 22.5 m - 24 m



15-SWC-995: 24 m - 25.5 m



15-SWC-995: 25.5 m - 27 m



15-SWC-995: 27 m - 28.5 m



15-SWC-995: 28.5 m - 30 m



15-SWC-995: 30 m - 31.5 m



15-SWC-995: 31.5 m - 33 m



15-SWC-995: 33 m - 34.5 m



15-SWC-995: 34.5 m - 36 m



15-SWC-995: 36 m - 37.5 m



15-SWC-995: 37.5 m - 39 m



15-SWC-995: 39 m - 40.5 m



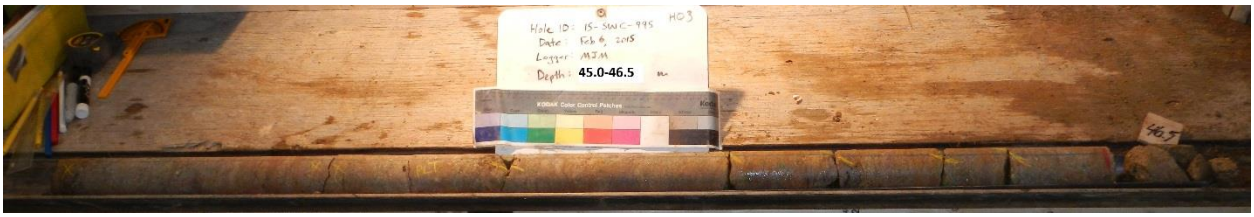
15-SWC-995: 40.5 m - 42 m



15-SWC-995: 42 m - 43.5 m



15-SWC-995: 43.5 m - 45 m



15-SWC-995: 45 m - 46.5 m



15-SWC-995: 46.5 m - 48 m



15-SWC-995: 48 m - 49.5 m



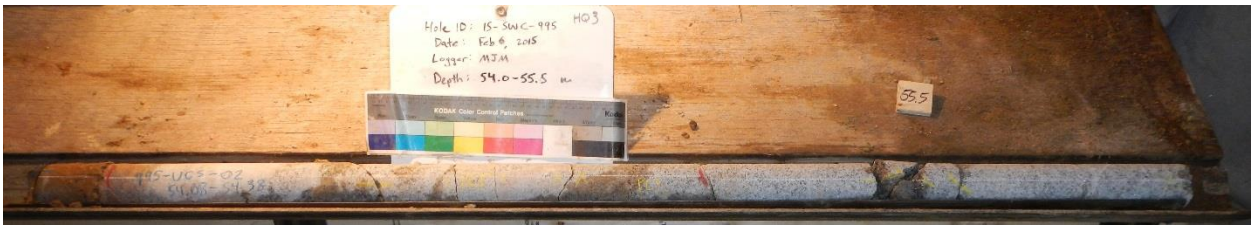
15-SWC-995: 49.5 m - 51 m



15-SWC-995: 51 m - 52.5 m



15-SWC-995: 52.5 m - 54 m



15-SWC-995: 54 m - 55.5 m



15-SWC-995: 55.5 m - 57 m



15-SWC-995: 57 m - 58.5 m



15-SWC-995: 58.5 m - 60 m



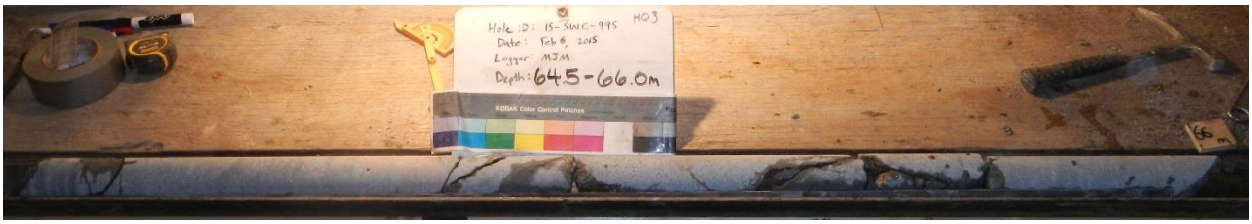
15-SWC-995: 60 m - 61.5 m



15-SWC-995: 61.5 m - 63 m



15-SWC-995: 63 m - 64.5 m



15-SWC-995: 64.5 m - 66 m



15-SWC-995: 66 m - 67.5 m



15-SWC-995: 67.5 m - 69 m



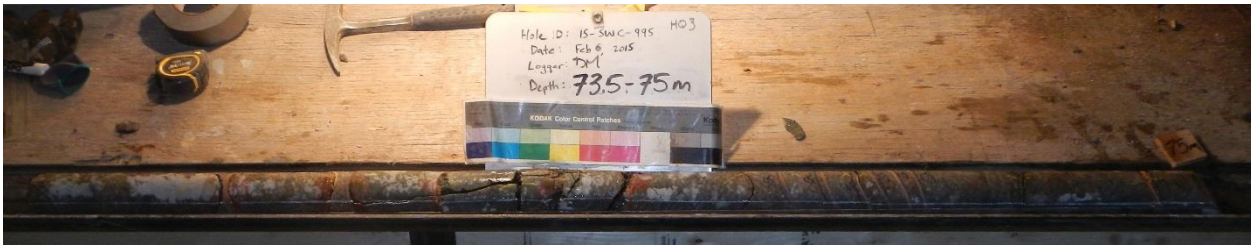
15-SWC-995: 69 m - 70.5 m



15-SWC-995: 70.5 m - 72 m



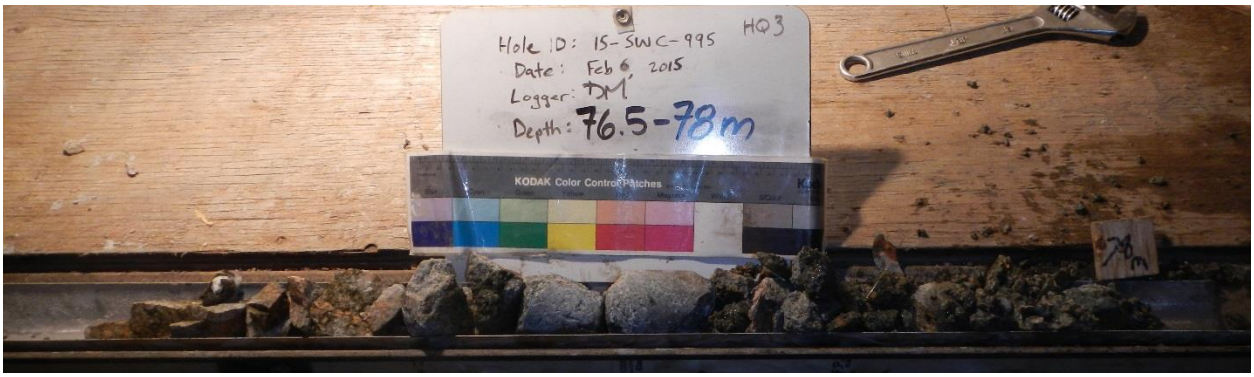
15-SWC-995: 72 m - 73.5 m



15-SWC-995: 73.5 m - 75 m



15-SWC-995: 75 m - 76.5 m



15-SWC-995: 76.5 m - 78 m



15-SWC-995: 78 m - 79.5 m



15-SWC-995: 79.5 m - 81 m



15-SWC-995: 81 m - 82.5 m



15-SWC-995: 82.5 m - 84 m



15-SWC-995: 84 m - 85.5 m



15-SWC-995: 85.5 m - 87 m



15-SWC-995: 87 m - 88.5 m



15-SWC-995: 88.5 m - 90 m



15-SWC-995: 90 m - 91.5 m



15-SWC-995: 91.5 m - 93 m



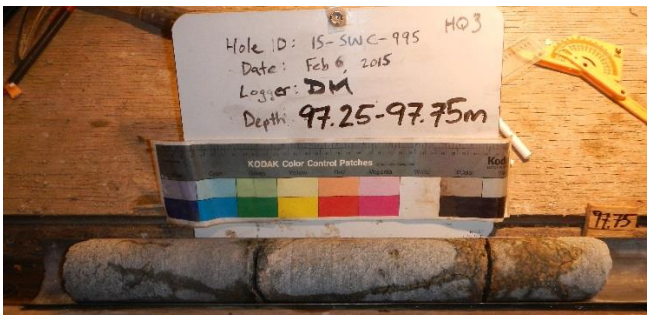
15-SWC-995: 93 m - 94.25 m



15-SWC-995: 94.25 m - 95.75 m



15-SWC-995: 95.75 m - 97.25 m



15-SWC-995: 97.25 m - 97.75 m



15-SWC-995: 97.75 m - 99.25 m



15-SWC-995: 99.25 m - 100.75 m



15-SWC-995: 100.75 m - 102.25 m



15-SWC-995: 102.25 m - 103.75 m



15-SWC-995: 103.75 m - 105 m



15-SWC-995: 105 m - 106.5 m



15-SWC-995: 106.5 m - 108 m



15-SWC-995: 108 m - 109.5 m



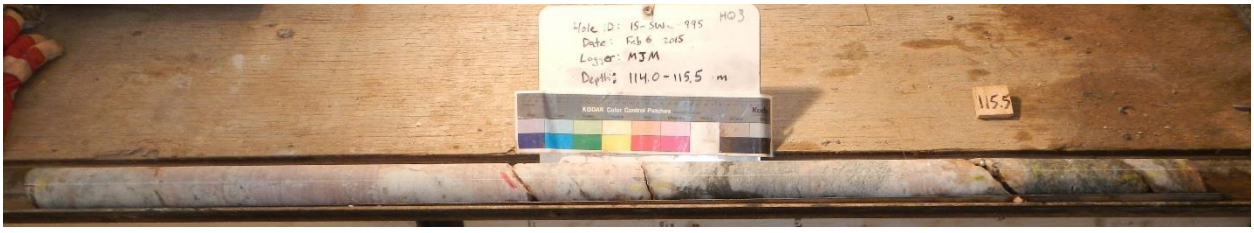
15-SWC-995: 109.5 m - 111 m



15-SWC-995: 111 m - 112.5 m



15-SWC-995: 112.5 m - 114 m



15-SWC-995: 114 m - 115.5 m



15-SWC-995: 115.5 m - 117 m



15-SWC-995: 117 m - 118.5 m



15-SWC-995: 118.5 m - 120 m



15-SWC-995: 120 m - 121.5 m



15-SWC-995: 121.5 m - 123 m



15-SWC-995: 123 m - 124.5 m



15-SWC-995: 124.5 m - 126 m



15-SWC-995: 126 m - 127.5 m



15-SWC-995: 127.5 m - 129 m



15-SWC-995: 129 m - 130.5 m



15-SWC-995: 130.5 m - 132 m



15-SWC-995: 132 m - 133.5 m



15-SWC-995: 133.5 m - 135 m



15-SWC-995: 135 m - 136.5 m



15-SWC-995: 136.5 m - 138 m



15-SWC-995: 138 m - 139.5 m



15-SWC-995: 139.5 m - 141 m



15-SWC-995: 141 m - 142.5 m



15-SWC-995: 142.5 m - 144 m



15-SWC-995: 144 m - 145.5 m



15-SWC-995: 145.5 m - 147 m



15-SWC-995: 147 m - 148.5 m



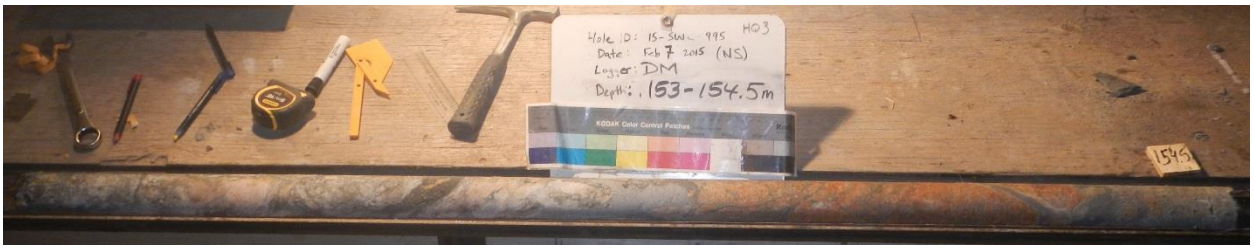
15-SWC-995: 148.5 m - 150 m



15-SWC-995: 150 m - 151.5 m



15-SWC-995: 151.5 m - 153 m



15-SWC-995: 153 m - 154.5 m



15-SWC-995: 154.5 m - 156 m



15-SWC-995: 156 m - 157.5 m



15-SWC-995: 157.5 m - 159 m



15-SWC-995: 159 m - 160.5 m



15-SWC-995: 160.5 m - 162 m



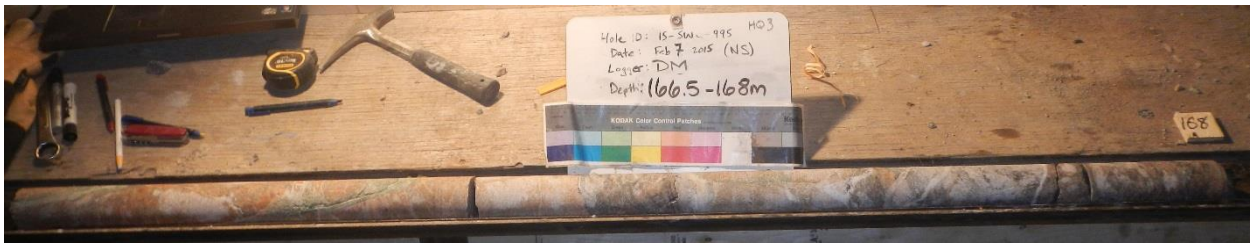
15-SWC-995: 162 m - 163.5 m



15-SWC-995: 163.5 m - 165 m



15-SWC-995: 165 m - 166.5 m



15-SWC-995: 166.5 m - 168 m



15-SWC-995: 168 m - 169.5 m



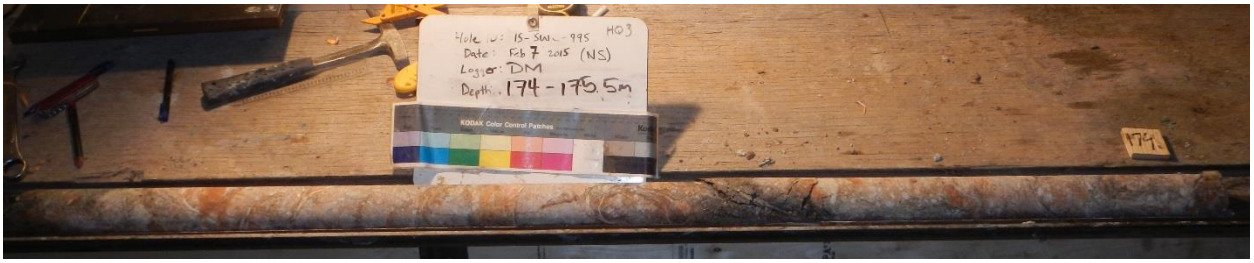
15-SWC-995: 169.5 m - 171 m



15-SWC-995: 171 m - 172.5 m



15-SWC-995: 172.5 m - 174 m



15-SWC-995: 174 m - 175.5 m



15-SWC-995: 175.5 m - 177 m



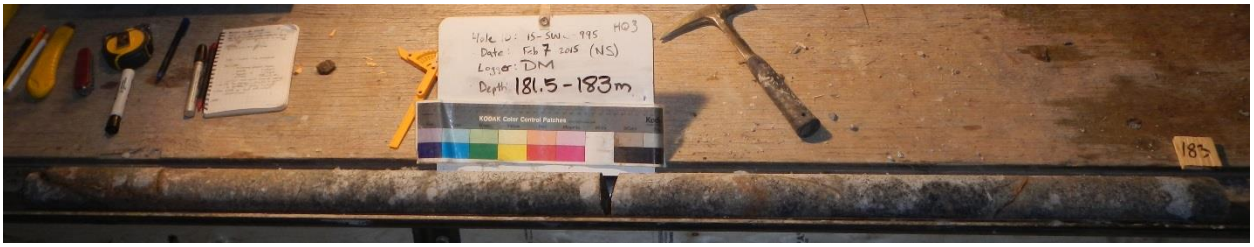
15-SWC-995: 177 m - 178.5 m



15-SWC-995: 178.5 m - 180 m



15-SWC-995: 180 m - 181.5 m



15-SWC-995: 181.5 m - 183 m



15-SWC-995: 183 m - 184.5 m



15-SWC-995: 184.5 m - 186 m



15-SWC-995: 186 m - 187.5 m



15-SWC-995: 189 m - 190.5 m



15-SWC-995: 190.5 m - 192 m



15-SWC-995: 192 m - 193.5 m



15-SWC-995: 193.5 m - 195 m



15-SWC-995: 195 m - 196.5 m



15-SWC-995: 196.5 m - 198 m



15-SWC-995: 198 m - 199.5 m



15-SWC-995: 199.5 m - 201 m



15-SWC-995: 201 m - 202.5 m



15-SWC-995: 202.5 m - 204 m



15-SWC-995: 204 m - 205.5 m



15-SWC-995: 205.5 m - 207 m



15-SWC-995: 207 m - 208.2 m



15-SWC-995: 208.2 m - 209.7 m



15-SWC-995: 209.7 m - 211.3 m



15-SWC-995: 211.3 m - 212.8 m



15-SWC-995: 212.8 m - 214.4 m



15-SWC-995: 214.4 m - 216 m



15-SWC-995: 216 m - 217.5 m



15-SWC-995: 217.5 m - 219 m



15-SWC-995: 219 m - 220.5 m



15-SWC-995: 220.5 m - 221.8 m



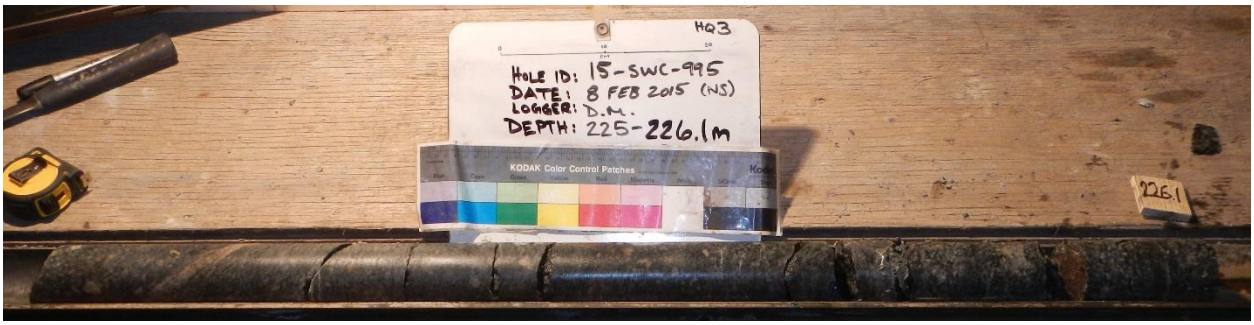
15-SWC-995: 221.8 m - 223.3 m



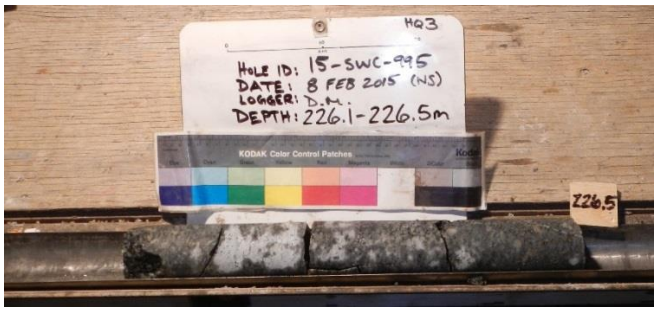
15-SWC-995: 223.3 m - 223.7 m



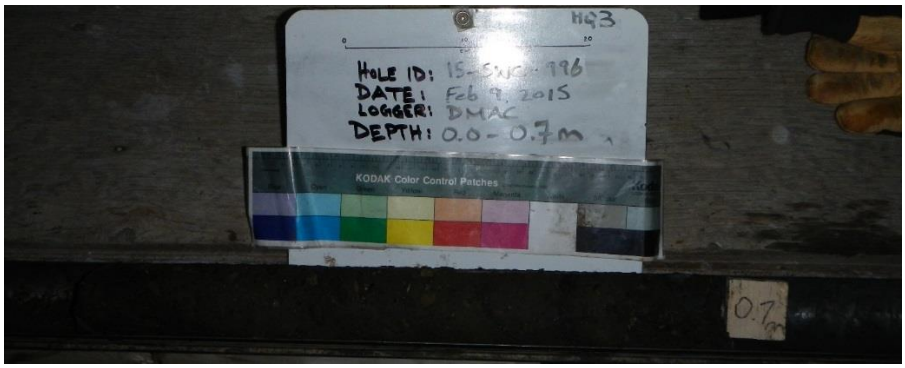
15-SWC-995: 223.7 m - 225 m



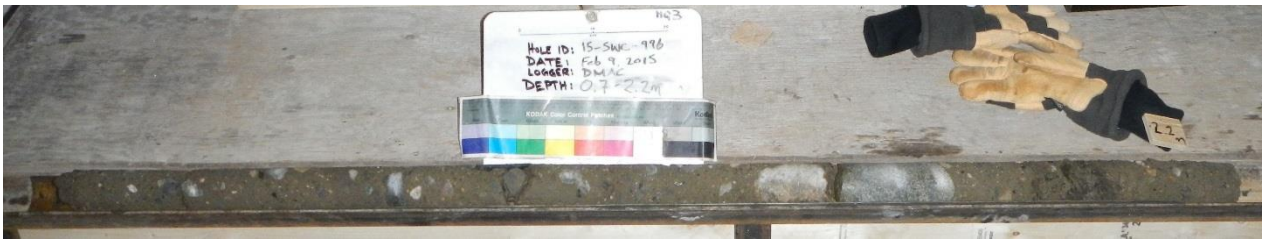
15-SWC-995: 225 m - 226.1 m



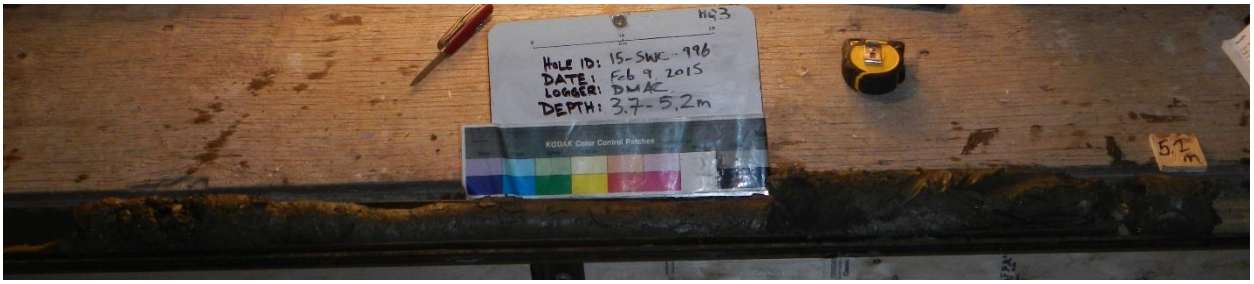
15-SWC-995: 226.1 m - 226.5 m



15-SWC-996: 0 m - 0.7 m



15-SWC-996: 0.7 m - 2.2 m



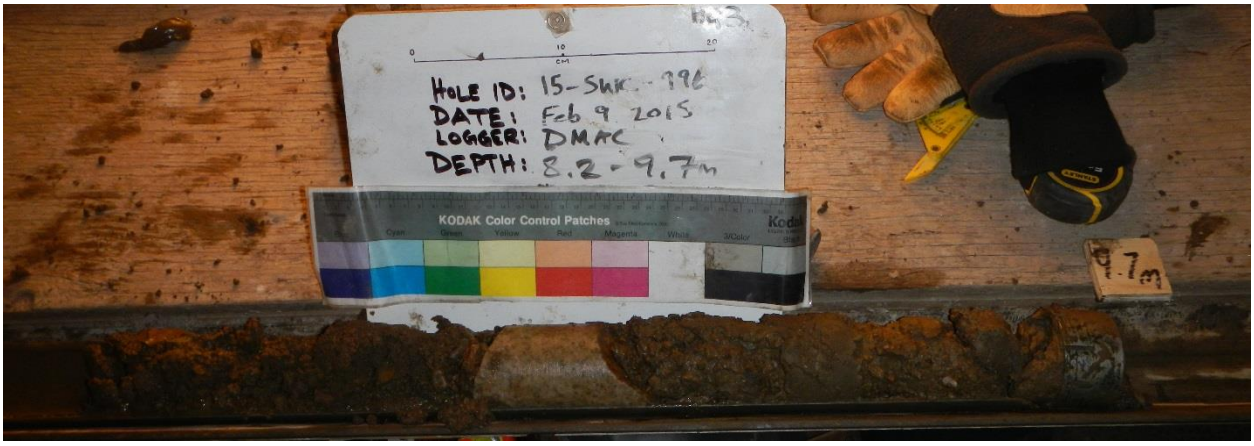
15-SWC-996: 3.7 m - 5.2 m



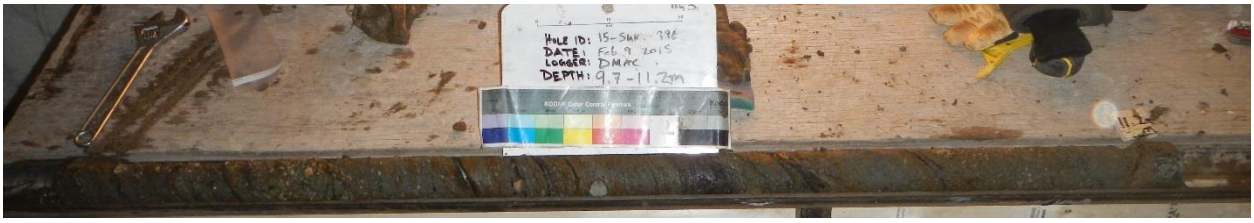
15-SWC-996: 5.2 m - 6.7 m



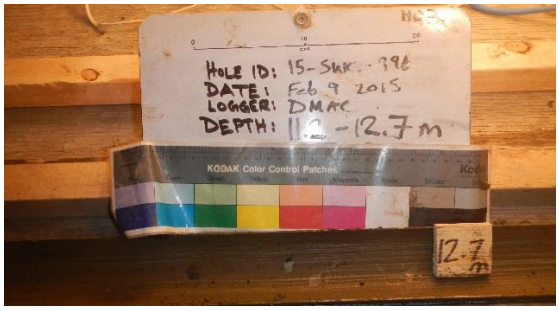
15-SWC-996: 6.7 m - 8.2 m



15-SWC-996: 8.2 m - 9.7 m



15-SWC-996: 9.7 m - 11.2 m



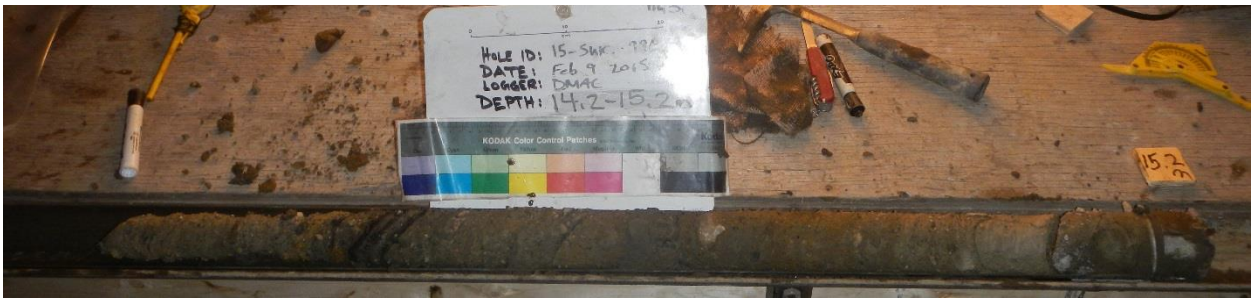
15-SWC-996: 11.2 m - 12.7 m



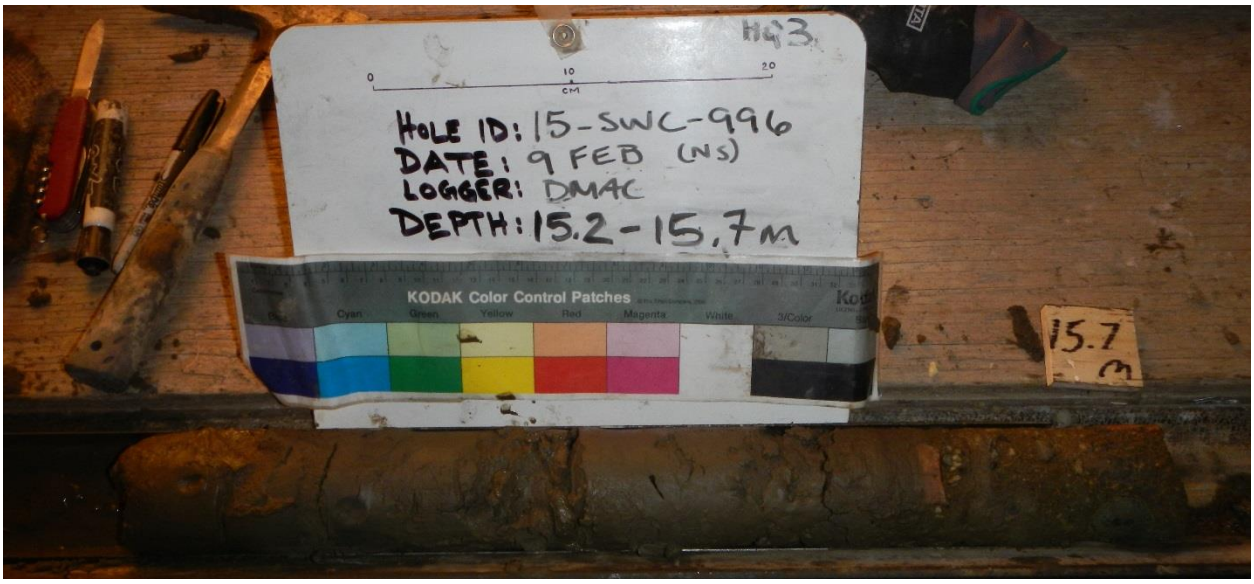
15-SWC-996: 12.7 m - 14.2 m



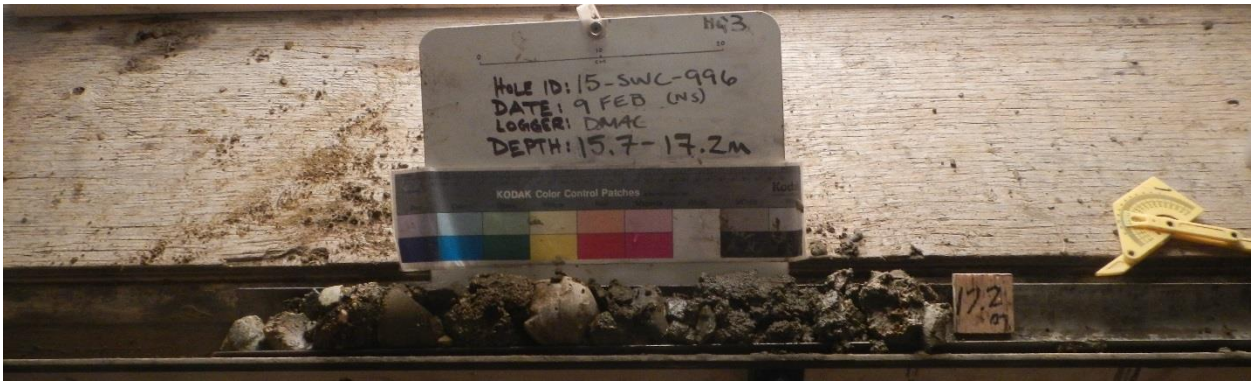
15-SWC-996: 13.6 m



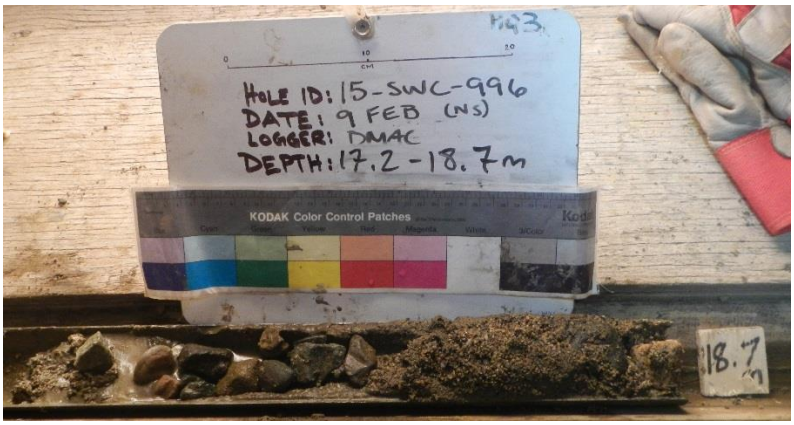
15-SWC-996: 14.2 m - 15.2 m



15-SWC-996: 15.2 m - 15.7 m



15-SWC-996: 15.7 m - 17.2 m



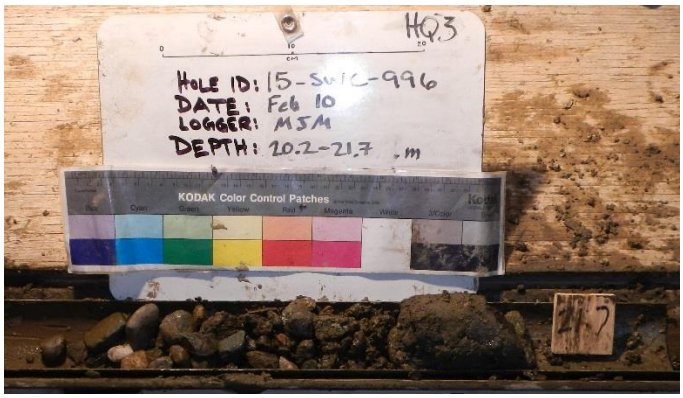
15-SWC-996: 17.2 m - 18.7 m



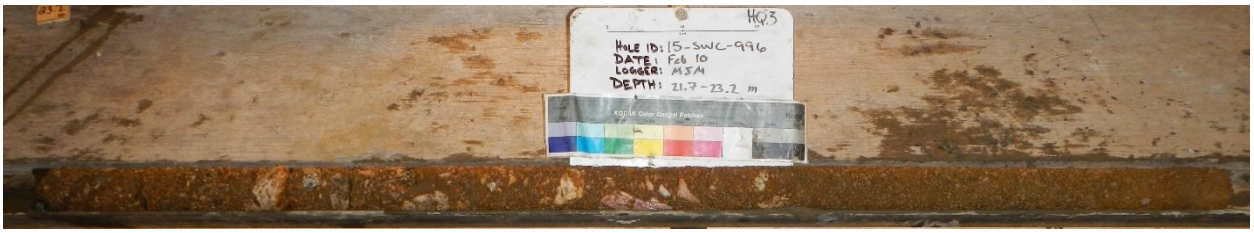
15-SWC-996: 18.7 m - 18.9 m



15-SWC-996: 18.9 m - 20.2 m



15-SWC-996: 20.2 m - 21.7 m



15-SWC-996: 21.7 m - 23.2 m



15-SWC-996: 23.2 m - 24.7 m



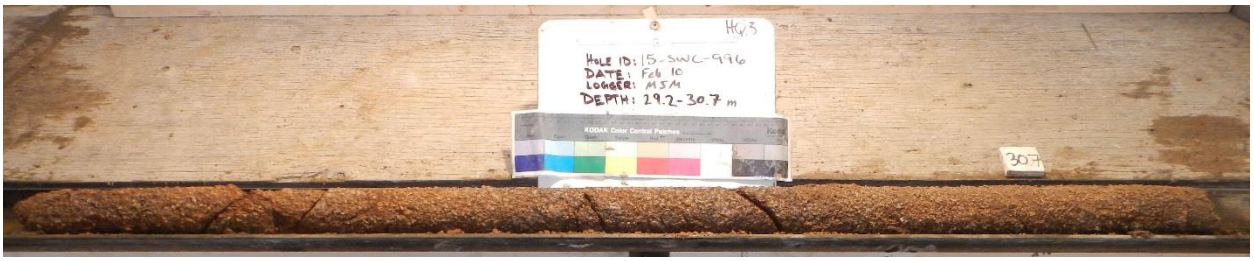
15-SWC-996: 24.7 m - 26.2 m



15-SWC-996: 26.2 m - 27.7 m



15-SWC-996: 27.7 m - 29.2 m



15-SWC-996: 29.2 m - 30.7 m



15-SWC-996: 30.7 m - 32.2 m



15-SWC-996: 32.2 m - 33.7 m



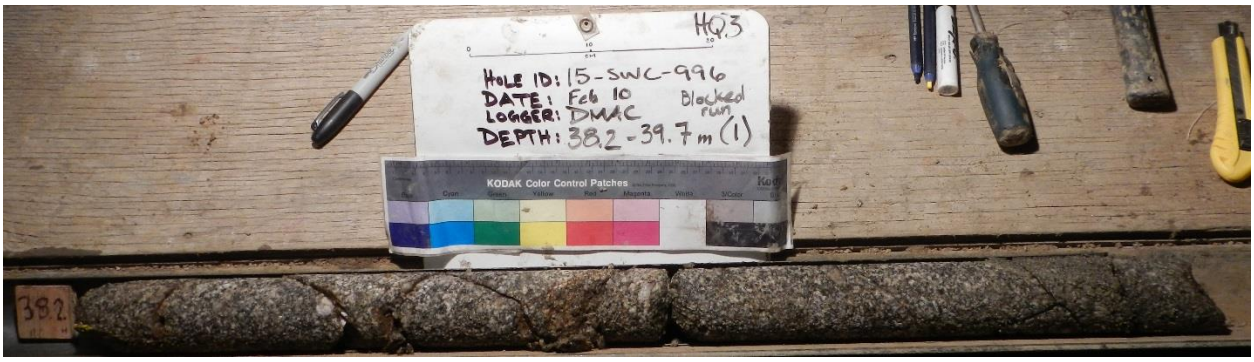
15-SWC-996: 33.7 m - 35.2 m



15-SWC-996: 35.2 m - 36.7 m



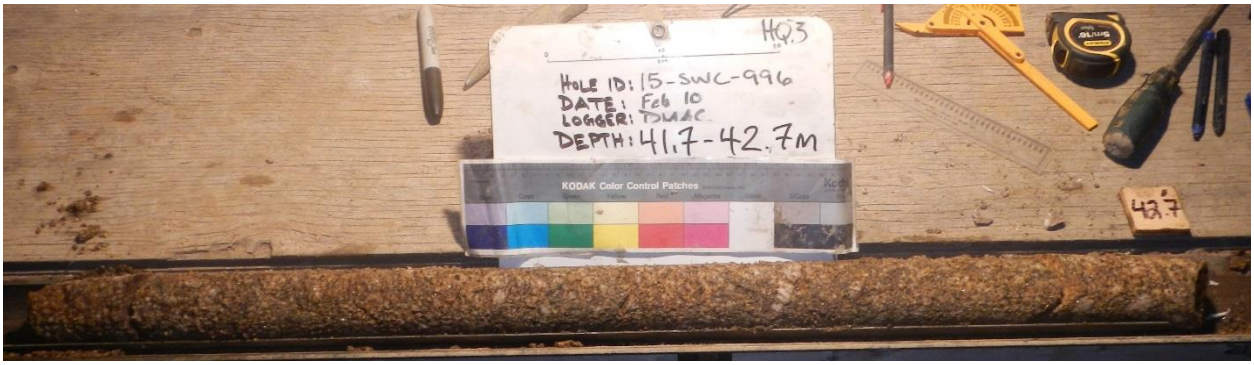
15-SWC-996: 36.7 m - 38.2 m



15-SWC-996: 38.2 m – 39.7 m



15-SWC-996: 39.7 m - 41.7 m



15-SWC-996: 41.7 m - 42.7 m



15-SWC-996: 42.7 m - 44.2 m



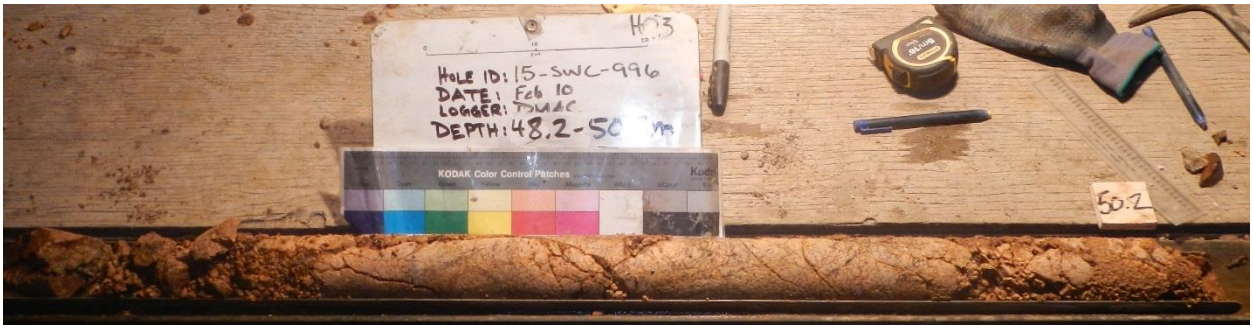
15-SWC-996: 44.2 m - 45.7 m



15-SWC-996: 45.7 m - 47.2 m



15-SWC-996: 47.2 m - 48.7 m



15-SWC-996: 48.7 m - 50.2 m



15-SWC-996: 50.2 m - 51.7 m



15-SWC-996: 51.7 m - 53.2 m



15-SWC-996: 53.2 m - 54.7 m



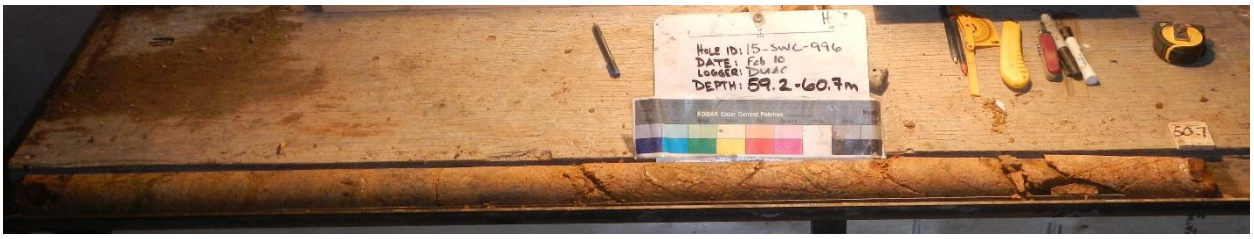
15-SWC-996: 54.7 m - 56.2 m



15-SWC-996: 56.2 m - 57.7 m



15-SWC-996: 57.7 m - 59.2 m



15-SWC-996: 59.2 m - 60.7 m



15-SWC-996: 60.7 m - 62.2 m



15-SWC-996: 62.2 m - 63.7 m



15-SWC-996: 63.7 m - 65.2 m



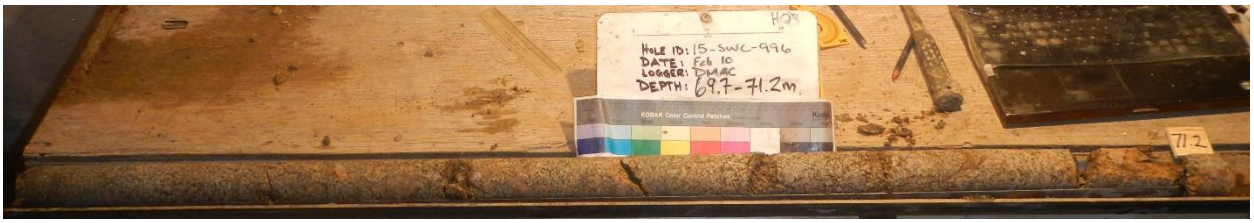
15-SWC-996: 65.2 m - 66.7 m



15-SWC-996: 66.7 m - 68.2 m



15-SWC-996: 68.2 m - 69.7 m



15-SWC-996: 69.7 m - 71.2 m



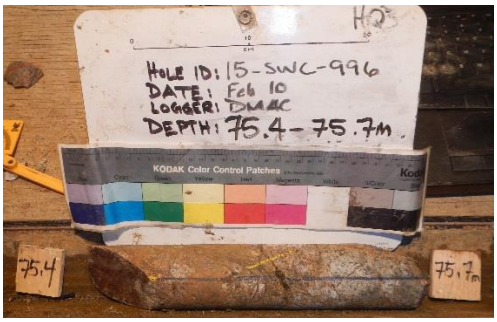
15-SWC-996: 71.2 m - 72.7 m



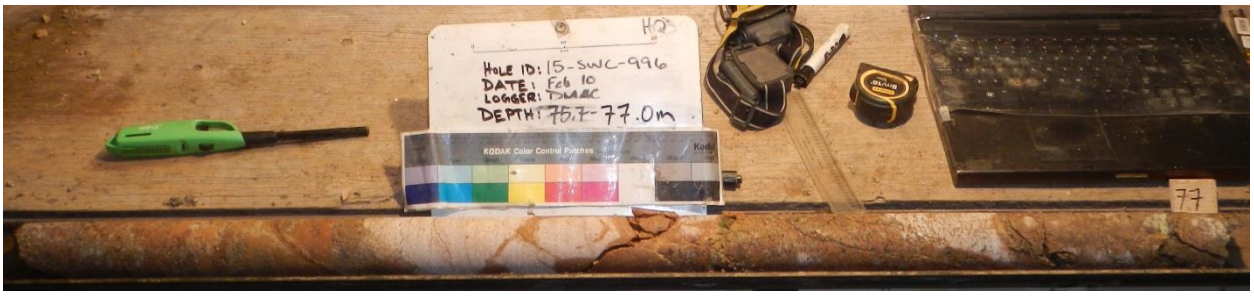
15-SWC-996: 72.7 m - 73.9 m



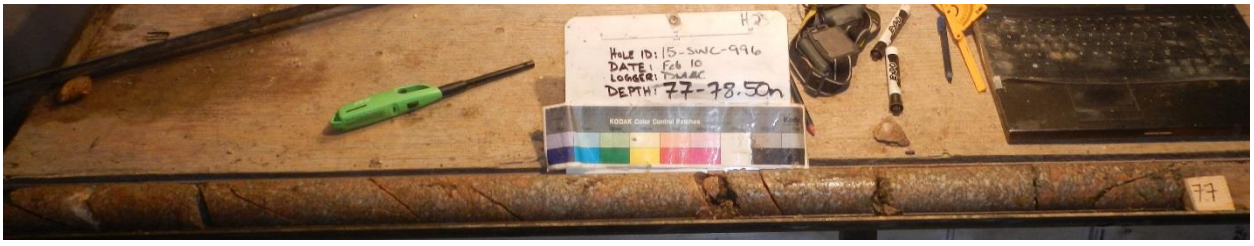
15-SWC-996: 73.9 m - 75.4 m



15-SWC-996: 75.4 m - 75.7 m



15-SWC-996: 75.7 m - 77 m



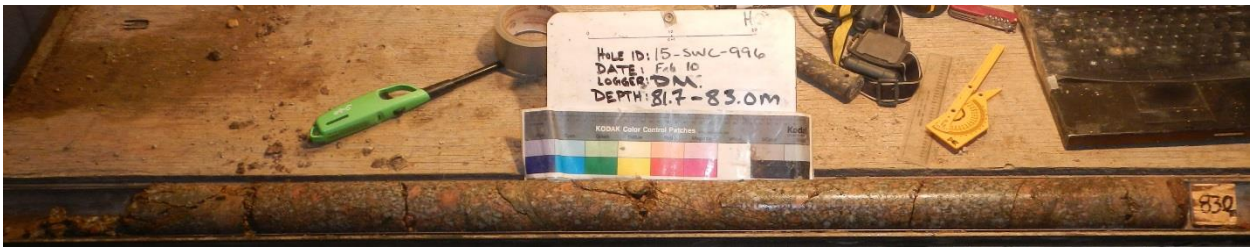
15-SWC-996: 77 m - 78.5 m



15-SWC-996: 78.5 m - 80.1 m



15-SWC-996: 80.1 m - 81.7 m



15-SWC-996: 81.7 m - 83 m



15-SWC-996: 83 m - 84.7 m



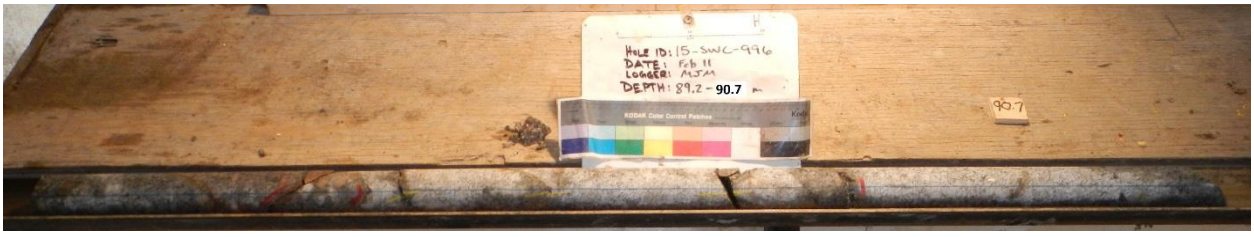
15-SWC-996: 84.7 m - 86.2 m



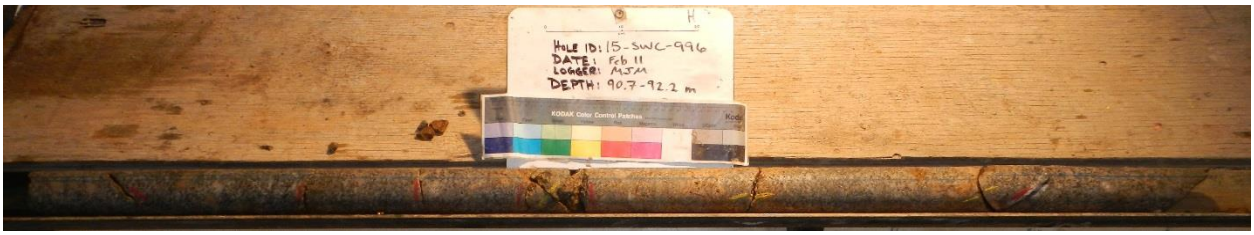
15-SWC-996: 86.2 m - 87.7 m



15-SWC-996: 87.7 m - 89.2 m



15-SWC-996: 89.2 m - 90.7 m



15-SWC-996: 90.7 m - 92.2 m



15-SWC-996: 92.2 m - 93.7 m



15-SWC-996: 93.7 m - 95.2 m



15-SWC-996: 95.2 m - 96.7 m



15-SWC-996: 96.7 m - 98.2 m



15-SWC-996: 98.2 m - 99.7 m



15-SWC-996: 99.7 m - 101.2 m



15-SWC-996: 101.2 m - 102.7 m



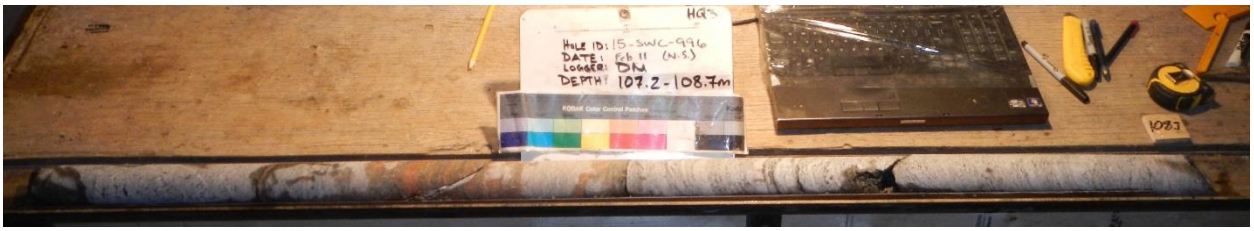
15-SWC-996: 102.7 m - 104.2 m



15-SWC-996: 104.2 m - 105.7 m



15-SWC-996: 105.7 m - 107.2 m



15-SWC-996: 107.2 m - 108.7 m



15-SWC-996: 108.7 m - 110.2 m



15-SWC-996: 110.2 m - 111.7 m



15-SWC-997: 0 m - 1.3 m



15-SWC-997: 1.3 m - 2.8 m



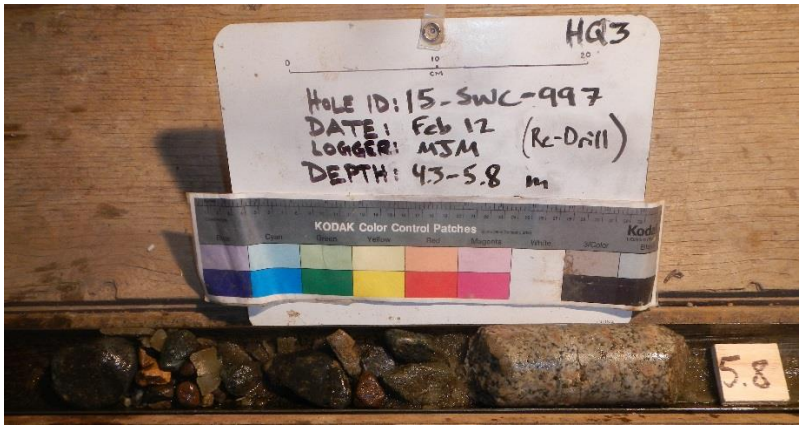
15-SWC-997: 2.8 m - 4.3 m



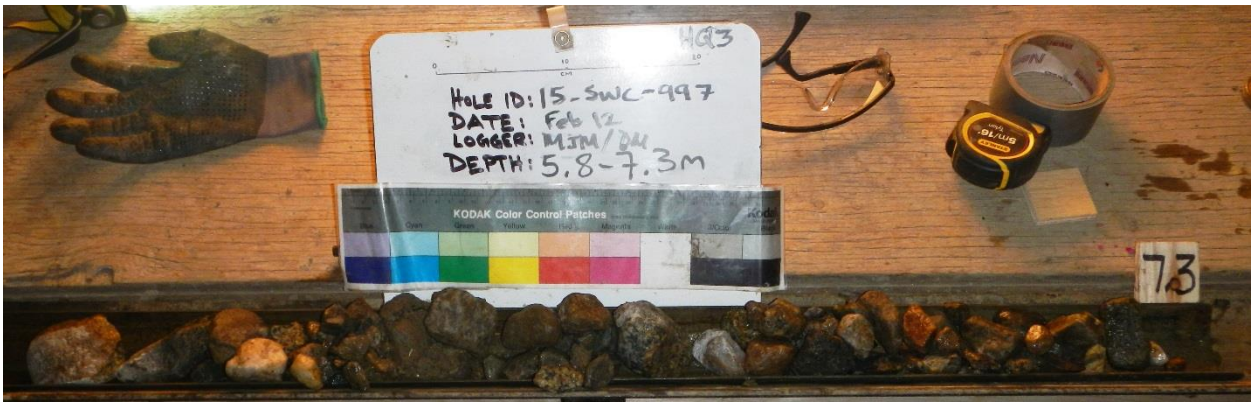
15-SWC-997: 0 m - 4.3 m re-drill



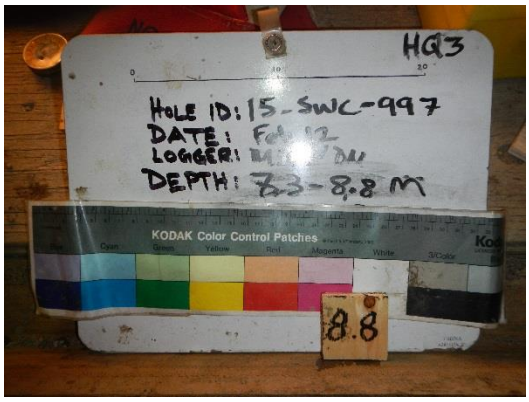
15-SWC-997: 4.3 m - 5.8 m



15-SWC-997: 4.3 m - 5.8 m re-drill



15-SWC-997: 5.8 m - 7.3 m



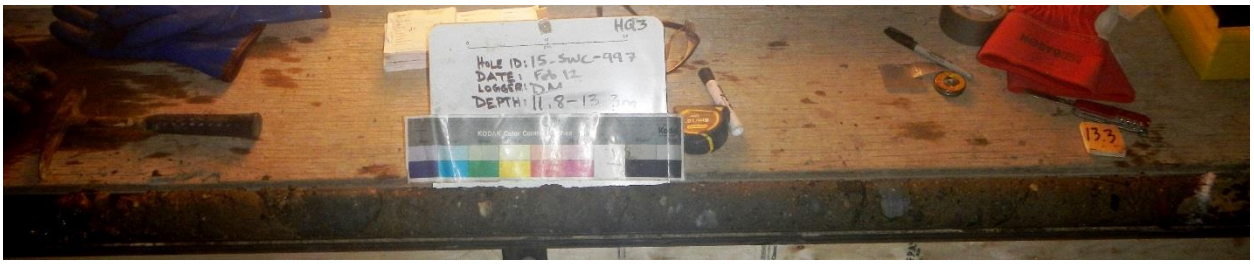
15-SWC-997: 7.3 m - 8.8 m



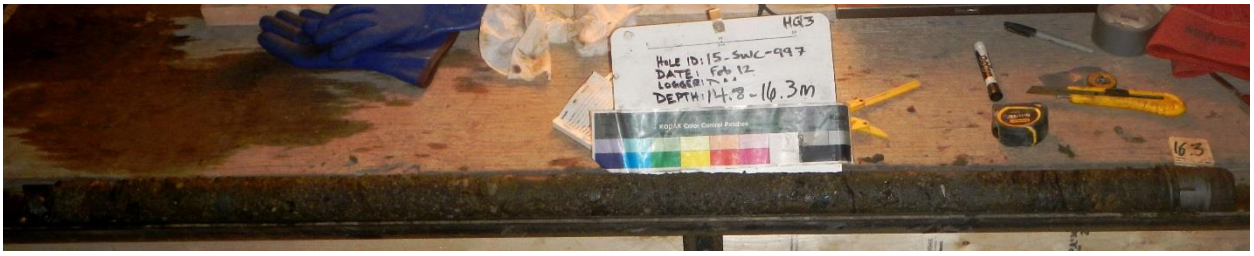
15-SWC-997: 8.8 m - 10.3 m



15-SWC-997: 10.3 m - 11.9 m



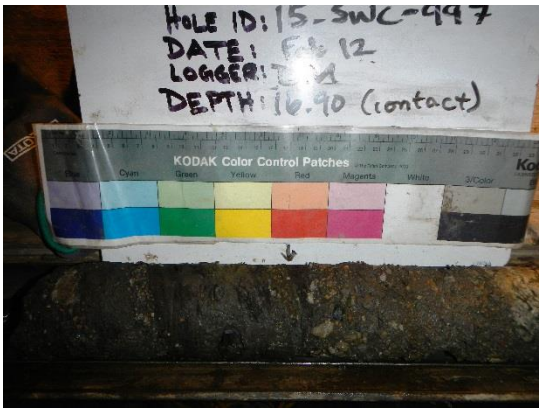
15-SWC-997: 11.8 m - 13.3 m



15-SWC-997: 14.8 m - 16.3 m



15-SWC-997: 16.3 m - 17.8 m



15-SWC-997: 16.9 m contact



15-SWC-997: 17.8 m - 19.3 m



15-SWC-997: 19.3 m - 20.8 m



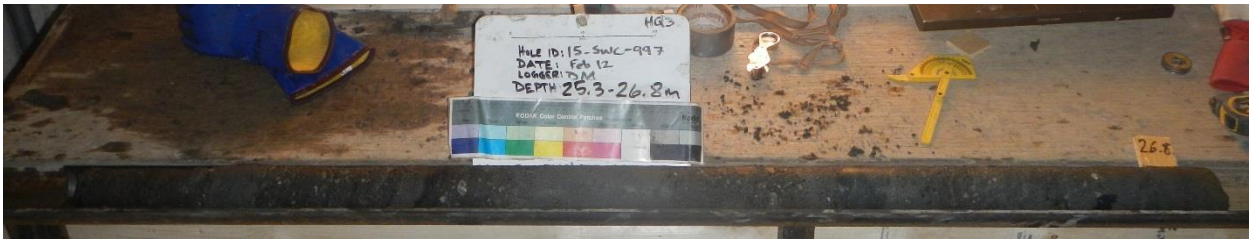
15-SWC-997: 20.8 m - 22.3 m



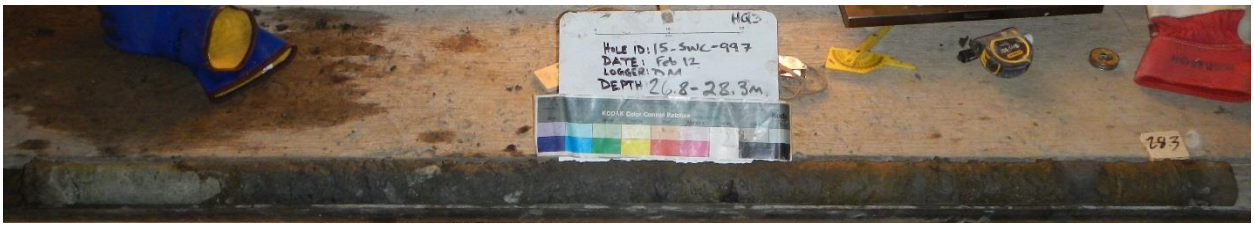
15-SWC-997: 22.3 m - 23.8 m



15-SWC-997: 23.8 m - 25.3 m



15-SWC-997: 25.3 m - 26.8 m



15-SWC-997: 26.8 m - 28.3 m



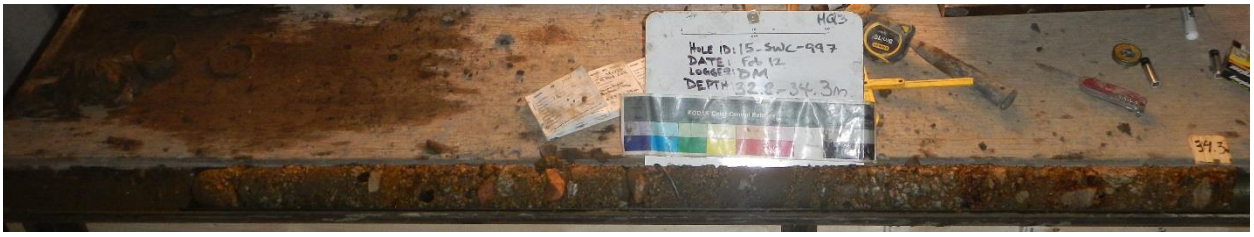
15-SWC-997: 28.3 m - 29.8 m



15-SWC-997: 29.8 m - 31.3 m



15-SWC-997: 31.3 m - 32.8 m



15-SWC-997: 32.8 m - 34.3 m



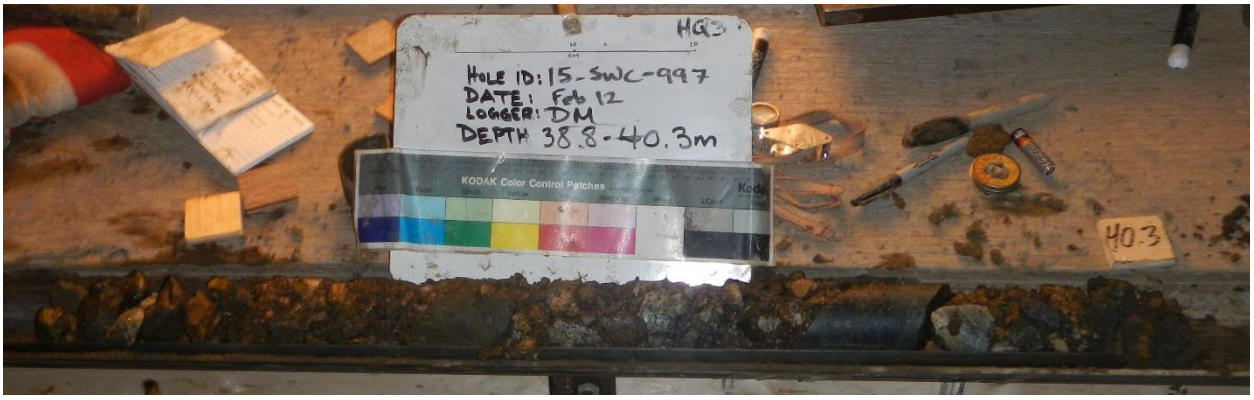
15-SWC-997: 34.3 m - 35.8 m



15-SWC-997: 35.8 m - 37.3 m



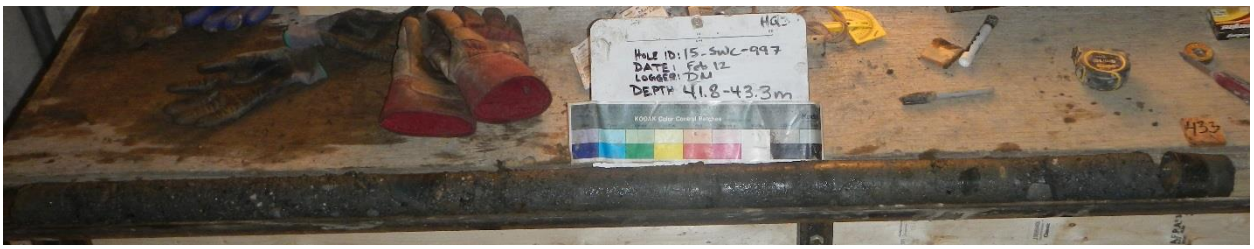
15-SWC-997: 37.3 m - 38.8 m



15-SWC-997: 38.8 m - 40.3 m



15-SWC-997: 40.3 m - 41.8 m



15-SWC-997: 41.8 m - 43.3 m



15-SWC-997: 43.3 m - 44.8 m



15-SWC-997: 44.8 m - 46.3 m



15-SWC-997: 46.3 m - 47.8 m



15-SWC-997: 47.8 m - 49.3 m



15-SWC-997: 49.3 m - 50.8 m



15-SWC-997: 50.8 m - 52.3 m



15-SWC-997: 52.3 m - 53.8 m



15-SWC-997: 53.8 m - 55.3 m



15-SWC-997: 55.3 m - 56.8 m



15-SWC-997: 56.8 m - 58.3 m



15-SWC-997: 58.3 m - 59.8 m



15-SWC-997: 59.8m – 61.3 m

Attachment 3: Soil Testing Results

MOISTURE CONTENT TEST RESULTS

ASTM D2216

Project:	SRK Minto Samples - Feb. 2015 Lab Testing	Sample No.:	
Project No.:	W14103546-01	Date Tested:	February 27, 2015
Client:	SRK Consulting (Canada) Inc.	Tested By:	AMT
Address:		Page:	1 of 1

B.H. Number	Sample Number	Moisture Content (%)	Visual Description of Soil
14-SWC-968	MM-101259	16.6	
14-SWC-970	MM-101262	27.6	
14-SWC-969	MM-101270	10.7	
14-SWC-969	MM-101273	26.3	
14-SWC-972	MM-101277	16.5	
14-SWC-972	MM-101280	31.7	
14-SWC-973	MM-101285	35.9	
14-SWC-973	MM-101288	13.2	
14-SWC-975	MM-101298	28.1	
14-SWC-979	MM-101317	26.5	
14-SWC-980	MM-101326	18.8	
15-SWC-995	58554	13.4	
15-SWC-996	58557	14.8	
15-SWC-996	58558	21.7	
15-SWC-996	58559	16.1	
15-SWC-997	160025	10.3	
15-SWC-997	58565	13.7	
15-SWC-997	58568	11.7	
15-SWC-997	58570	15.4	
15-SWC-997	58572	12.7	

Reviewed By: _____ P.Eng.

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TETRA TECH

FROZEN BULK RELATIVE DENSITY RESULTS

Project:	SRK Minto Samples - Feb. 2015 Lab Testing	Sample No.:	
Project No.:	W14103546-01	Date Tested:	27-Feb-15
Client:	SRK Consulting (Canada) Inc.	Tested By:	AMT
Address:		Page:	1 of 1

Bore Hole Number	Sample Number	Relative Density kg/m³	
15-SWC-997	160025	2317.63	
15-SWC-997	58568	2221.89	
15-SWC-997	58570	2224.24	
15-SWC-997	58572	2243.78	
15-SWC-997	58565	2353.06	
15-SWC-996	58557	2186.32	
15-SWC-996	58558	2026.10	
15-SWC-996	58559	2126.37	

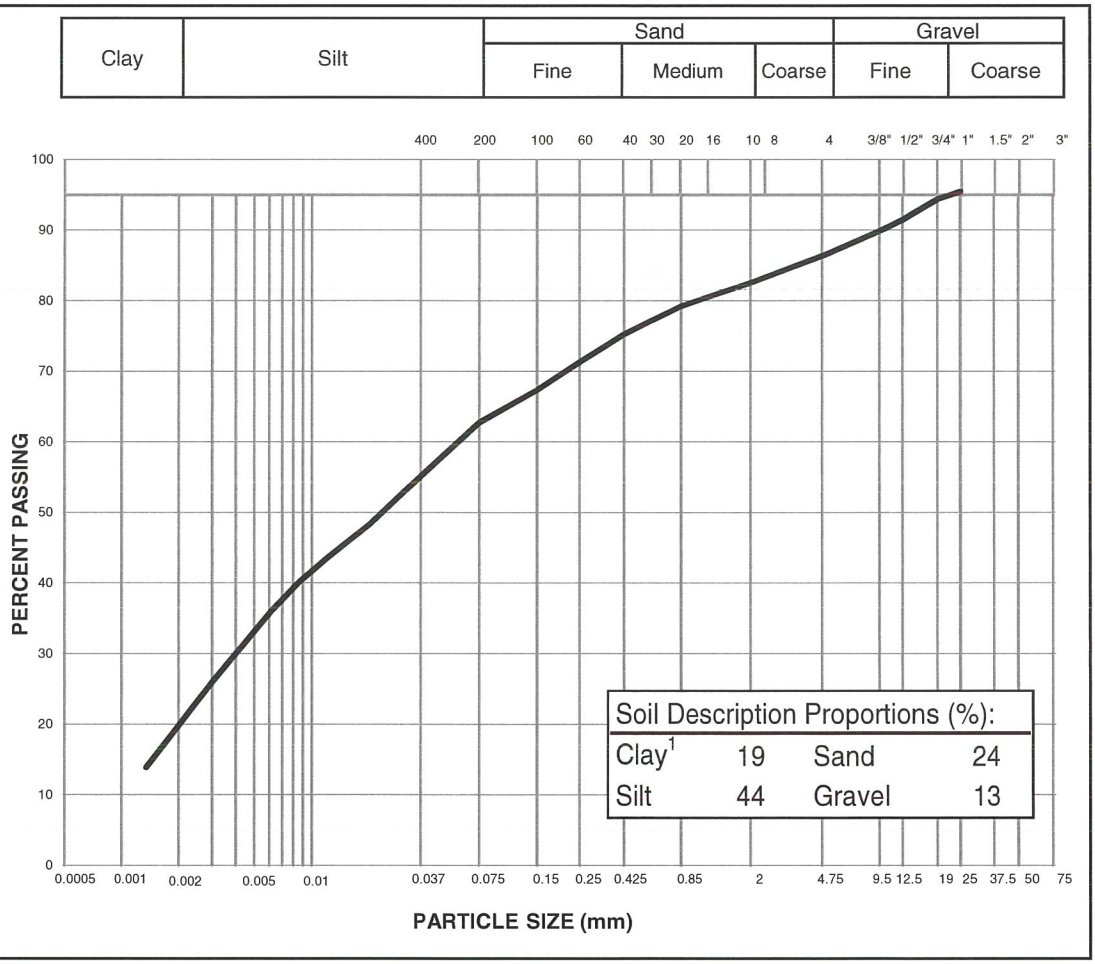
Reviewed By:

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	MM-101259
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	14-SWC-968
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	15.05 - 15.40 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 5, 2015 By: AMT	Date sampled:	November 1, 2014
Soil Description[2]:	SILT - sandy, some clay, some gravel	Sampled By:	Client
		USC Classification:	Cu: #N/A
Moisture Content:	16.6%		Cc: #N/A

Particle Size (mm)	Percent Passing
75	
50	
38	
25	95
19	94
12.5	91
10	90
5	87
2	83
0.85	79
0.425	75
0.25	71
0.15	67
0.075	62.7
0.0308	53.1
0.0199	48.2
0.0117	43.3
0.0085	40.0
0.0061	35.9
0.0030	26.1
0.0014	13.9



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

_____

Reviewed By: _____ C.E.T.

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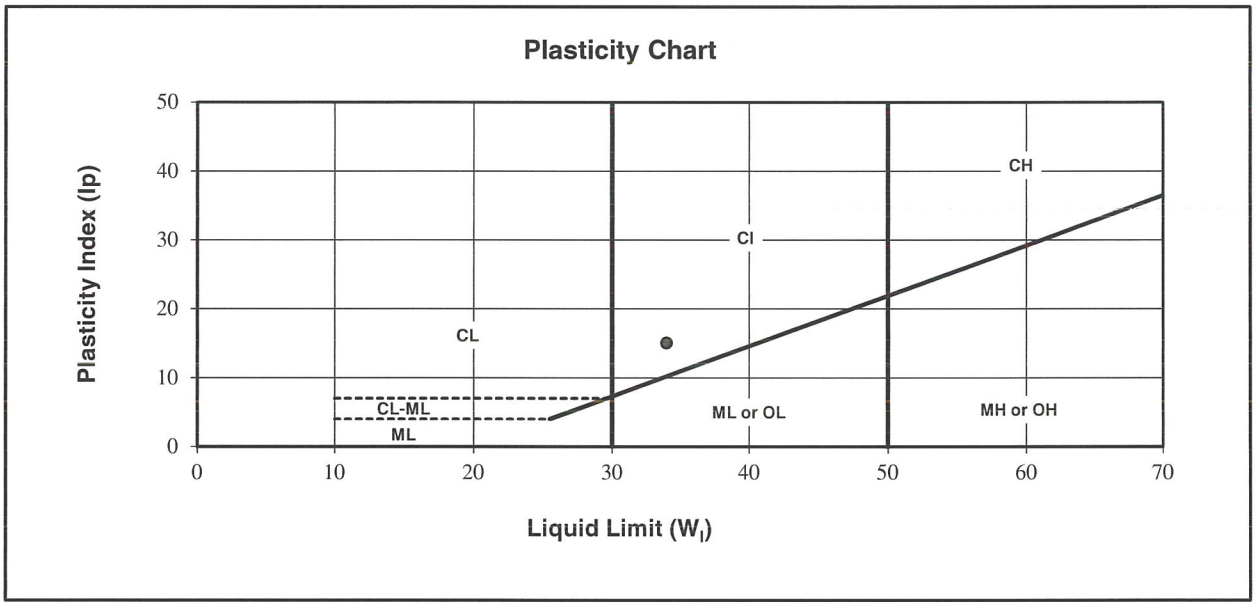
TETRA TECH

ATTERBERG LIMITS TEST REPORT

ASTM D4318

Project:	Minto February 2015 Lab Testing	Sample Number:	MM-101259
		Borehole Number:	14-SWC-968
Project No:	W14103546-01	Depth:	15.05 - 15.40 m
Client:	SRK Consulting (Canada) Inc.	Sampled By:	Client
Attention:	Murrary McGregor	Tested By:	AMT
Email:		Date Sampled:	November 1, 2014
		Date Tested:	March 6, 2015

Sample Description: SILT - sandy, some clay, some gravel



Liquid Limit ($W_{1)}$:	34	Natural Moisture (%)	16.6
Plastic Limit :	19	Soil Plasticity:	Medium
Plasticity Index (Ip) :	**15**	Mod.USCS Symbol:	CI

Remarks:

Reviewed By: C.E.T.

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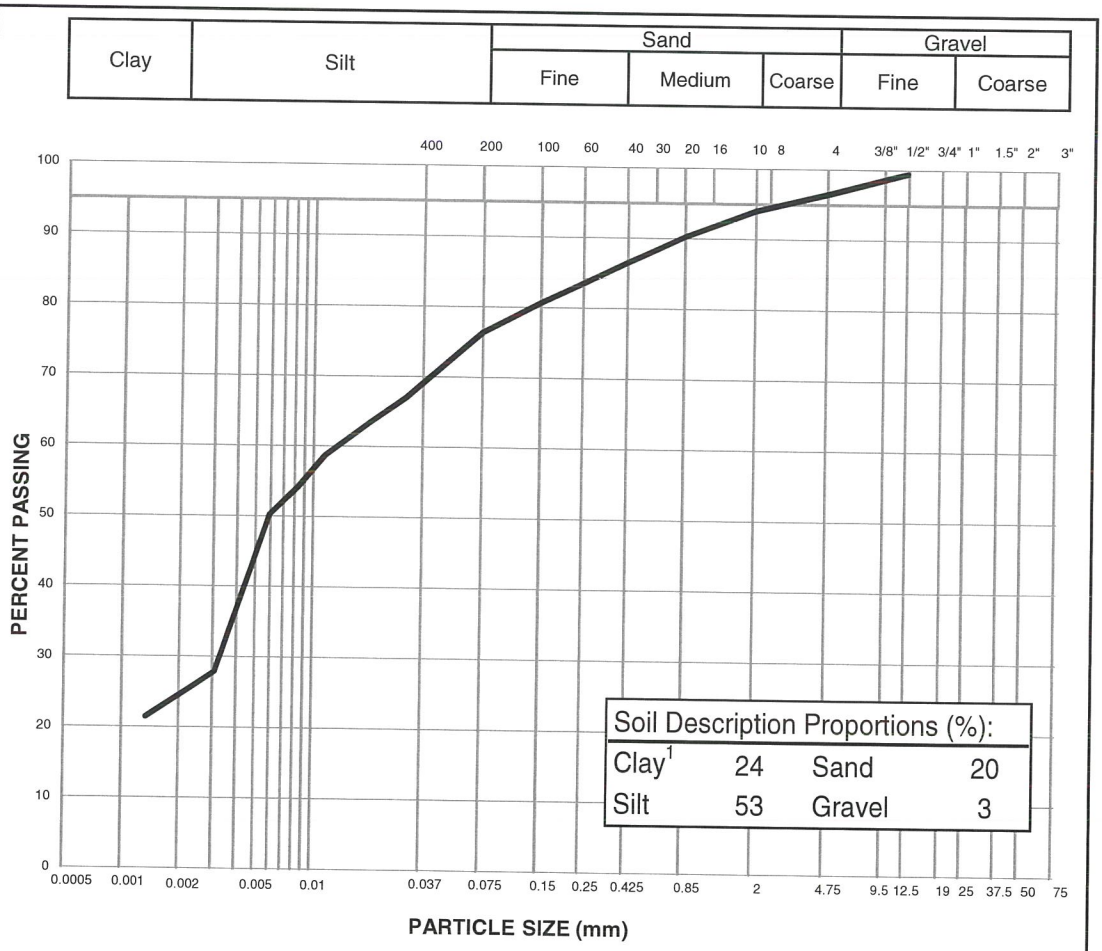
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	MM-101262
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	14-SWC-970
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	5.65 - 6.00 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 3, 2015 By: AMT	Date sampled:	November 2, 2014
Soil Description[2]:	SILT and CLAY - some sand, trace gravel	Sampled By:	Client
		USC Classification:	Cu: #N/A
Moisture Content:	27.6%		Cc: #N/A

Particle Size (mm)	Percent Passing
75	
50	
38	
25	
19	
12.5	99
10	99
5	97
2	94
0.85	90
0.425	87
0.25	84
0.15	81
0.075	76.4
0.0303	67.0
0.0194	63.3
0.0114	58.7
0.0082	54.0
0.0059	50.3
0.0031	27.9
0.0013	21.4



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

Reviewed By: _____ C.E.T.

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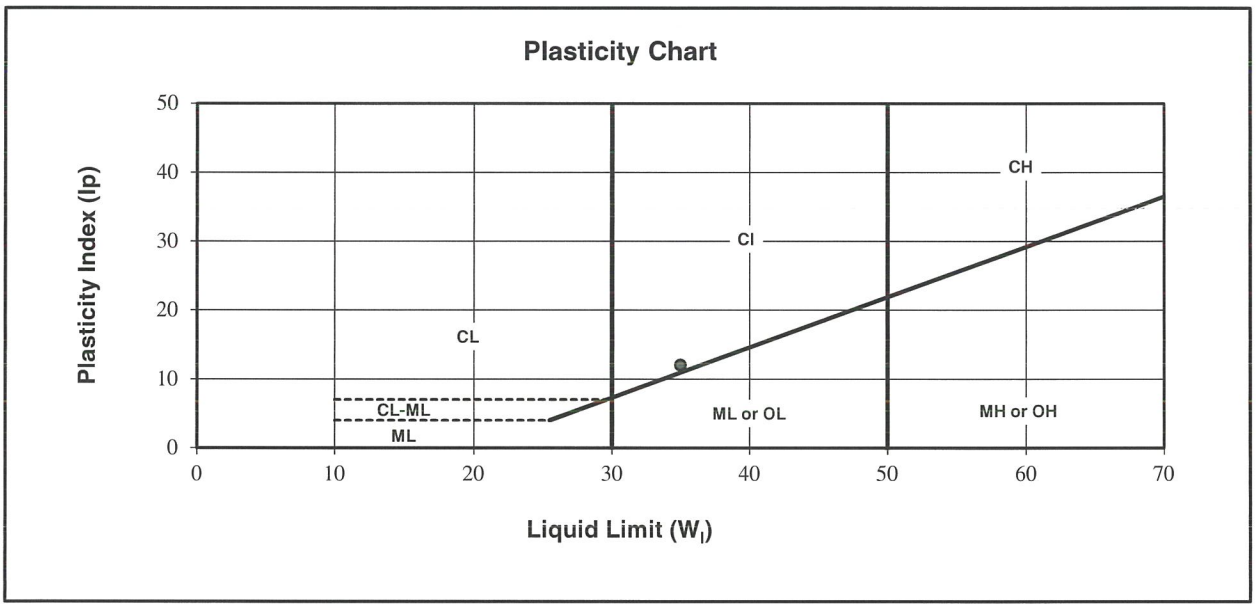
TETRA TECH

ATTERBERG LIMITS TEST REPORT

ASTM D4318

Project:	Minto February 2015 Lab Testing	Sample Number:	MM-101262
		Borehole Number:	14-SWC-967
Project No:	W14103546-01	Depth:	5.65 - 6.00 m
Client:	SRK Consulting (Canada) Inc.	Sampled By:	Client
		Tested By:	AMT
Attention:	Murrary McGregor	Date Sampled:	November 2, 2014
Email:		Date Tested:	March 6, 2015

Sample Description: SILT and CLAY - some sand, trace gravel



Liquid Limit ($W_{1)}$:	35	Natural Moisture (%)	16.6
Plastic Limit :	23	Soil Plasticity:	Medium
Plasticity Index (Ip) :	**12**	Mod.USCS Symbol:	CI

Remarks:

Reviewed By: C.E.T.

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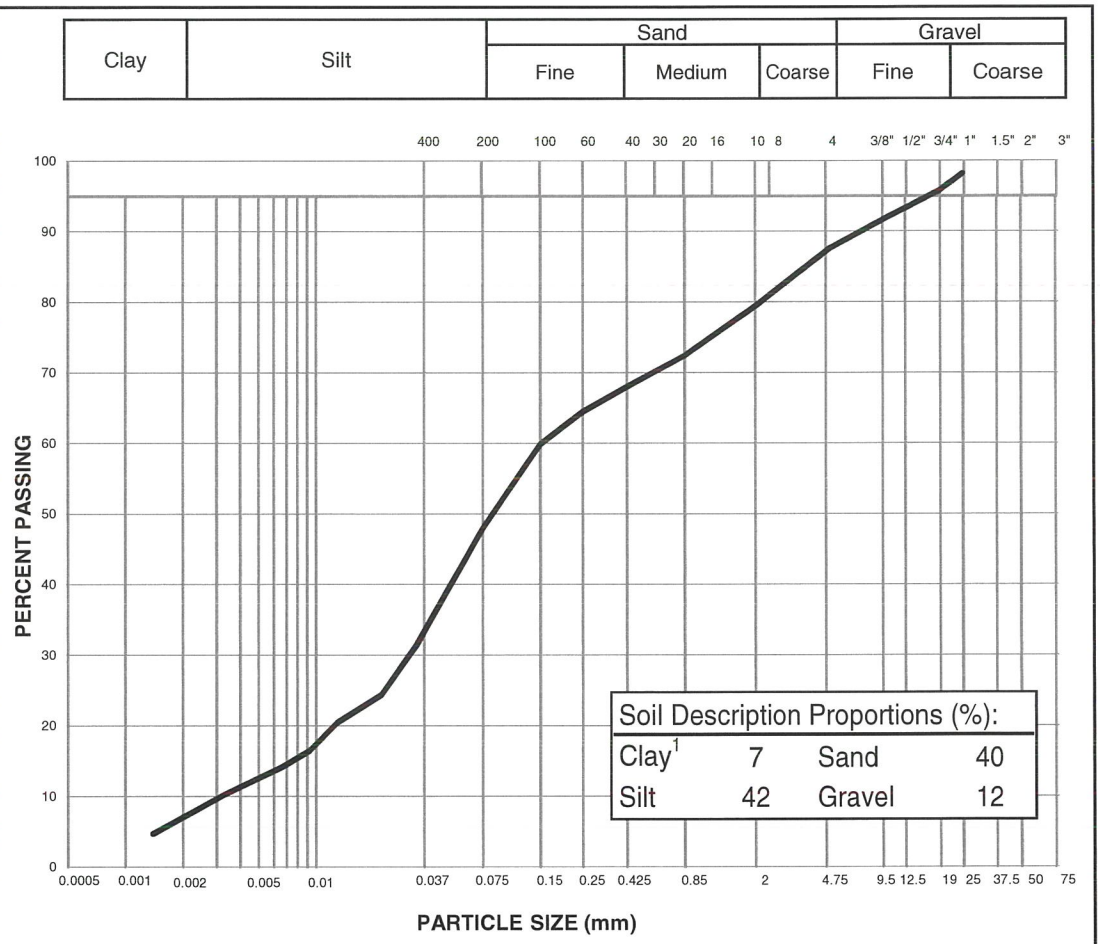
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	MM-101270
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	14-SWC-969
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	12.00 - 12.30 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 4, 2015 By: AMT	Date sampled:	November 2, 2014
Soil Description[2]:	SILT and SAND - some gravel, trace clay	Sampled By:	Client
		USC Classification:	Cu: 47.7
Moisture Content:	10.7%		Cc: 2.0

Particle Size (mm)	Percent Passing
75	
50	
38	
25	98
19	96
12.5	93
10	92
5	88
2	79
0.85	72
0.425	68
0.25	64
0.15	60
0.075	48.0
0.0339	31.4
0.0220	24.4
0.0129	20.4
0.0093	16.5
0.0066	14.1
0.0033	10.2
0.0014	4.7



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

Reviewed By: _____ C.E.T.

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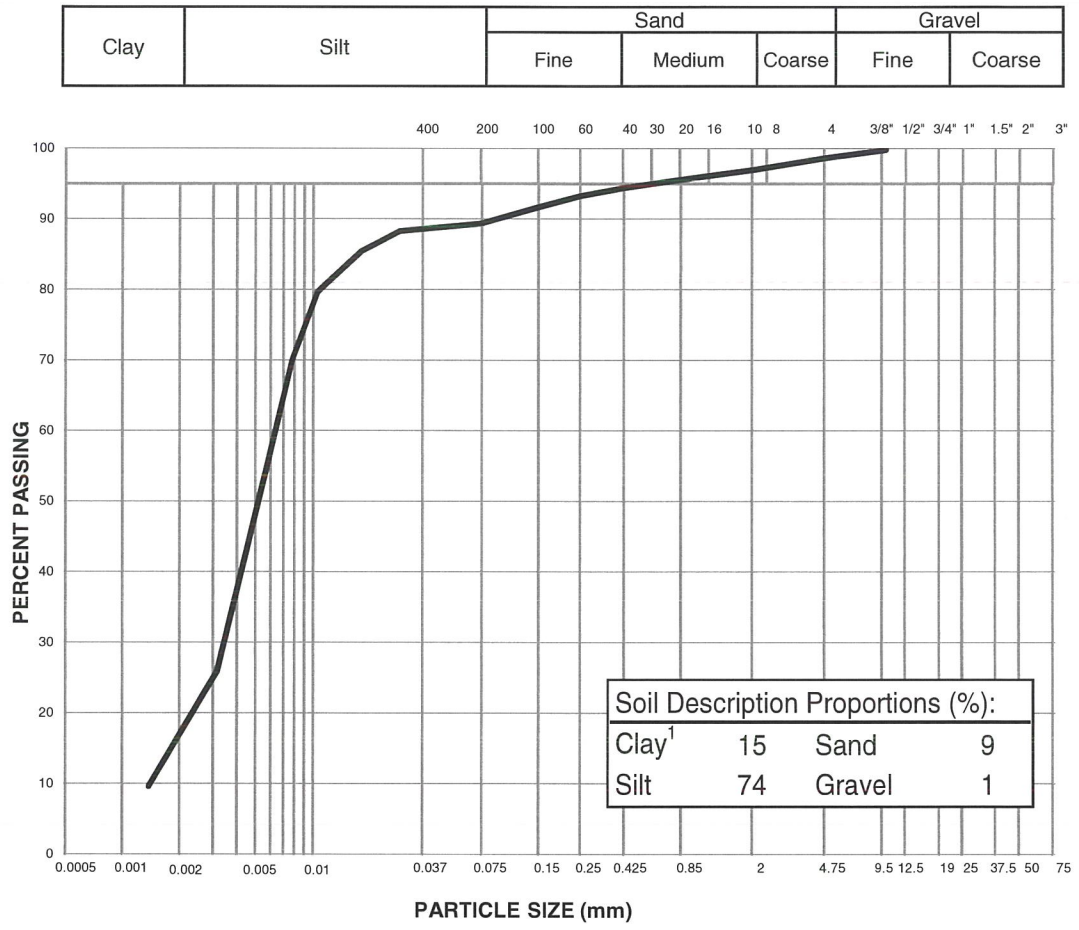
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	MM-101273
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	14-SWC-969
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	19.05 - 19.35 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 4, 2015 By: AMT	Date sampled:	November 2, 2014
Soil Description[2]:	SILT - some clay, trace sand, trace gravel	Sampled By:	Client
		USC Classification:	Cu: 4.5
Moisture Content:	26.3%		Cc: 1.4

Particle Size (mm)	Percent Passing
75	
50	
38	
25	
19	
12.5	
10	100
5	99
2	97
0.85	96
0.425	94
0.25	93
0.15	92
0.075	89.4
0.0282	88.3
0.0178	85.4
0.0106	79.7
0.0078	70.1
0.0058	55.7
0.0032	25.9
0.0014	9.6



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

_____

Reviewed By: _____ C.E.T.

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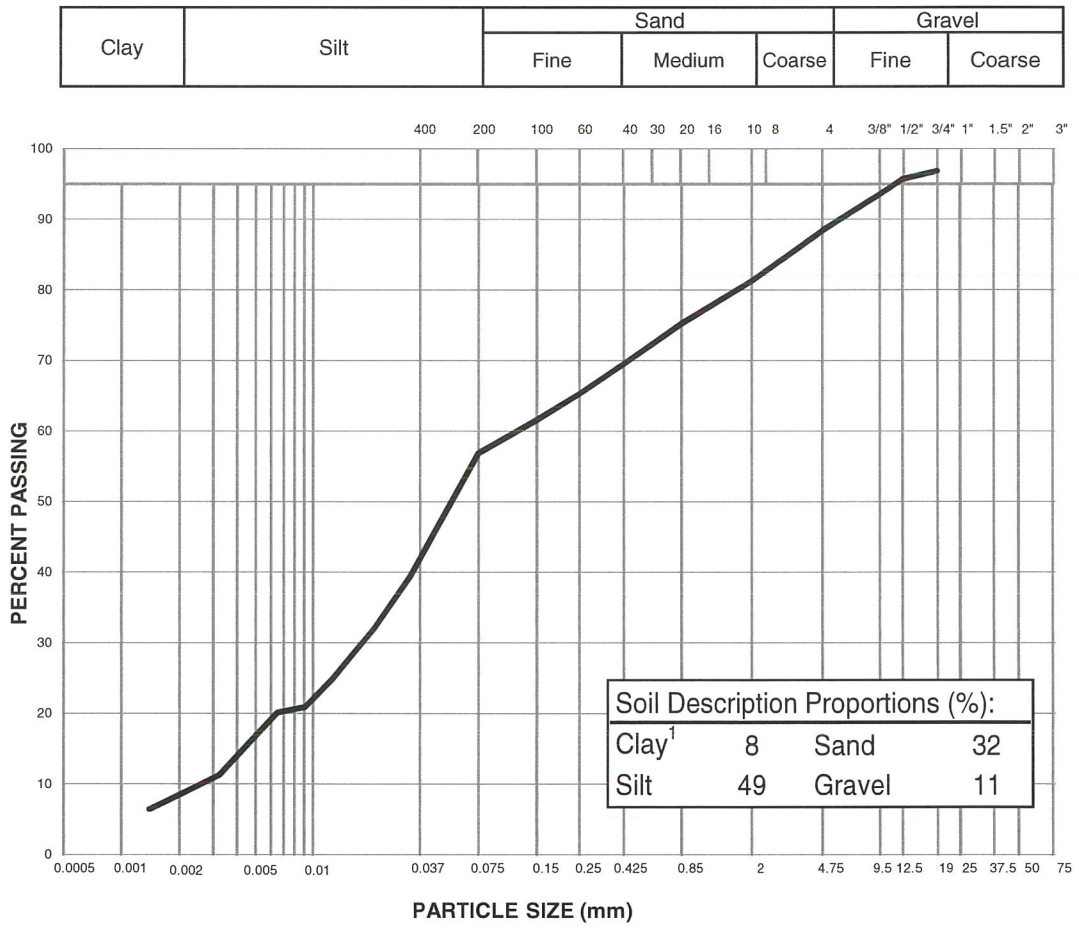
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	MM-101277
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	14-SWC-972
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	9.25 - 9.55 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 3, 2015 By: AMT	Date sampled:	November 4, 2014
Soil Description[2]:	SILT - sandy, some gravel, trace clay	Sampled By:	Client
		USC Classification:	Cu: 45.3
Moisture Content:	16.6%		Cc: 1.0

Particle Size (mm)	Percent Passing
75	
50	
38	
25	
19	97
12.5	96
10	94
5	89
2	81
0.85	75
0.425	70
0.25	65
0.15	62
0.075	56.9
0.0329	39.4
0.0214	32.2
0.0127	24.9
0.0091	20.9
0.0065	20.1
0.0032	11.3
0.0014	6.4



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

_____

Reviewed By: _____ C.E.T.

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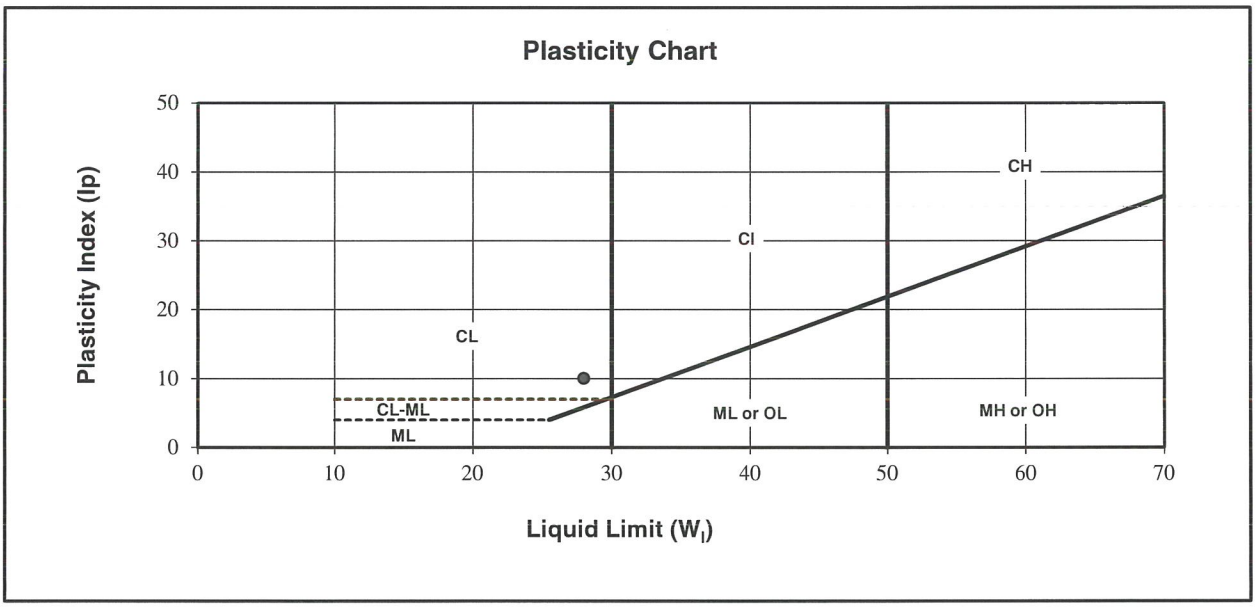
TETRA TECH

ATTERBERG LIMITS TEST REPORT

ASTM D4318

Project:	Minto February 2015 Lab Testing	Sample Number:	MM-101277
		Borehole Number:	14-SWC-272
Project No:	W14103546-01	Depth:	9.25 - 9.55 m
Client:	SRK Consulting (Canada) Inc.	Sampled By:	Client
		Tested By:	AMT
Attention:	Murrary McGregor	Date Sampled:	November 4, 2014
Email:		Date Tested:	March 5, 2015

Sample Description: SILT - sandy, some gravel, trace clay



Liquid Limit ($W_{1)}$:	28	Natural Moisture (%)	16.6
Plastic Limit :	18	Soil Plasticity:	Low to Medium
Plasticity Index (Ip) :	**10**	Mod.USCS Symbol:	CL-CI

Remarks:

Reviewed By: C.E.T.

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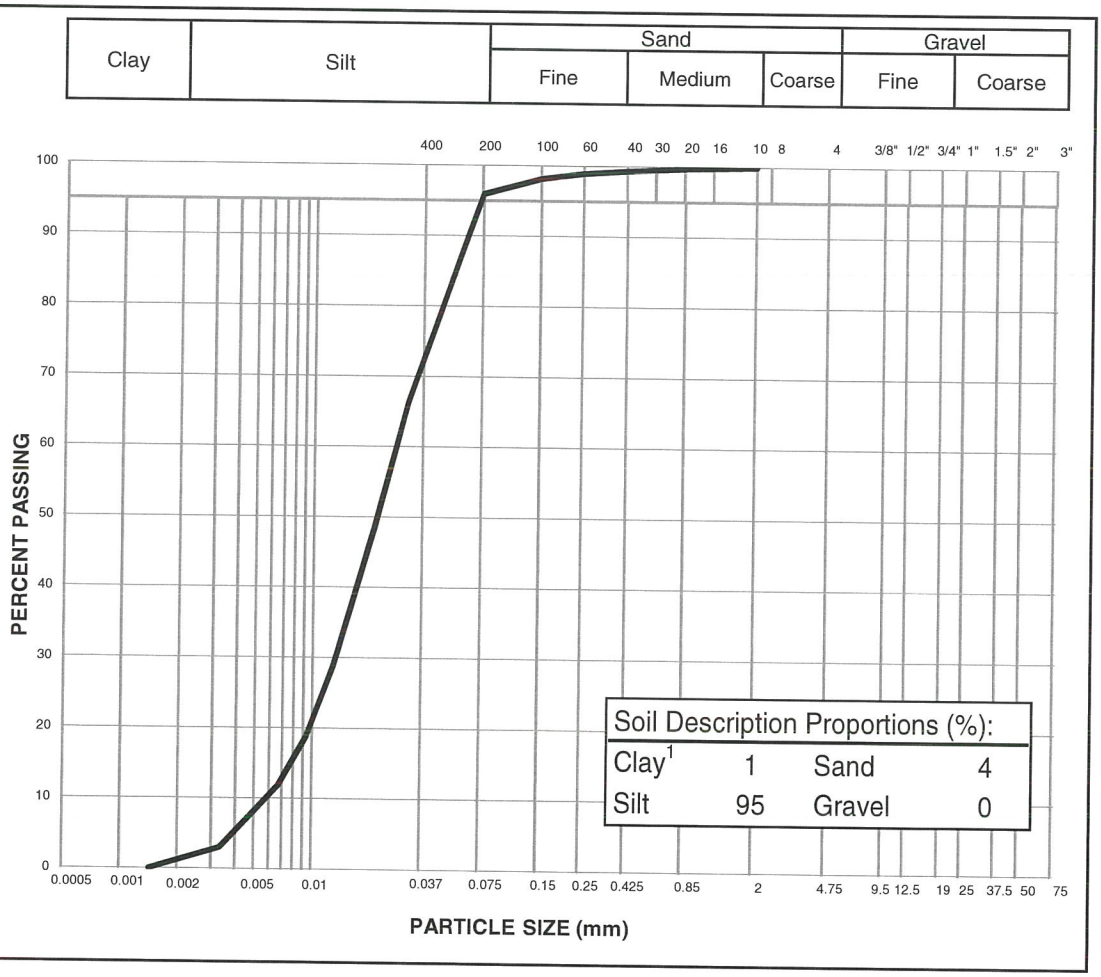
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	MM-101280
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	14-SWC-972
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	18.55 - 18.85 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 3, 2015 By: AMT	Date sampled:	November 4, 2014
Soil Description[2]:	SILT - trace sand, trace clay	Sampled By:	Client
		USC Classification:	Cu: 4.6
Moisture Content:	31.7%		Cc: 1.1

Particle Size (mm)	Percent Passing
75	
50	
38	
25	
19	
12.5	
10	
5	
2	100
0.85	100
0.425	99
0.25	99
0.15	98
0.075	96.0
0.0309	66.3
0.0208	48.5
0.0128	28.7
0.0093	18.8
0.0067	11.9
0.0033	3.0
0.0014	0.0



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

Reviewed By: _____ C.E.T.

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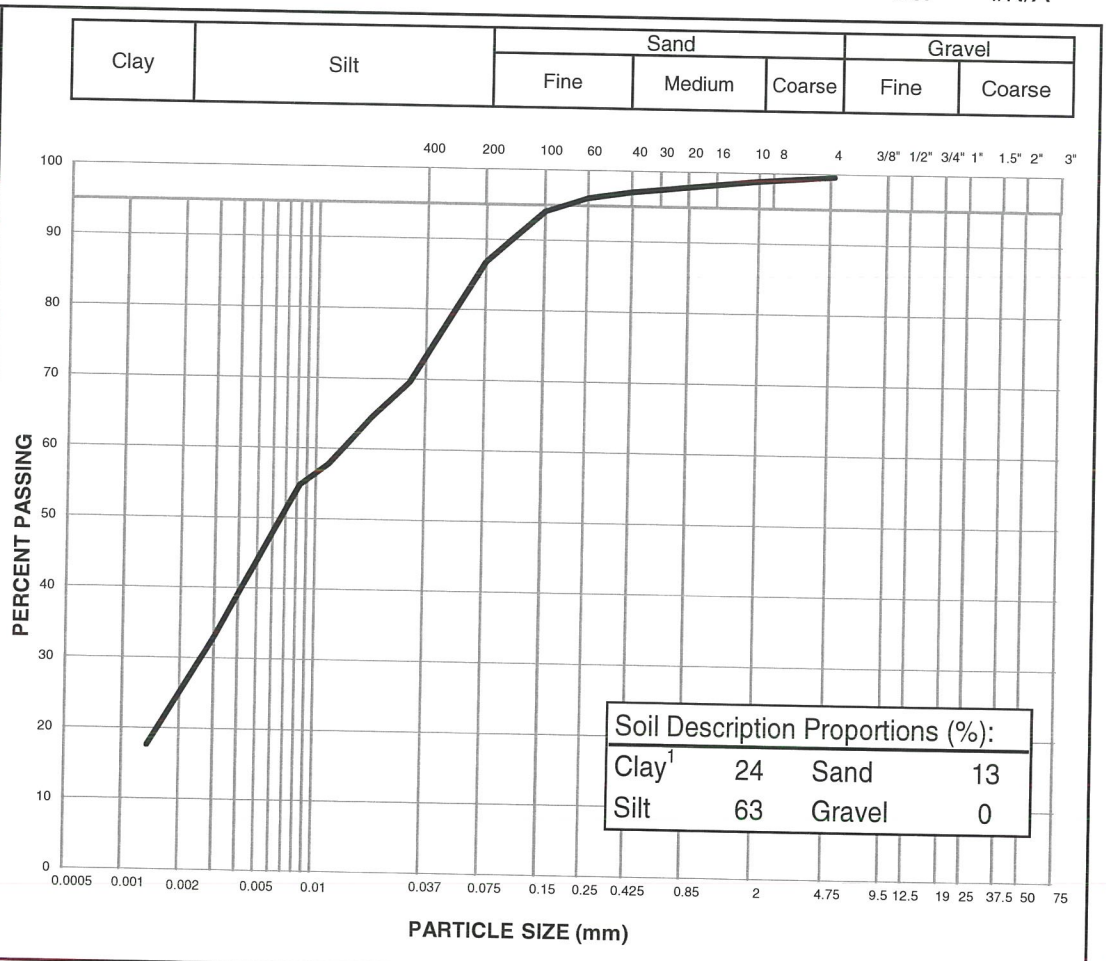
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	MM-101285
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	14-SWC-973
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	2.80 - 3.10 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 3, 2015 By: AMT	Date sampled:	November 6, 2014
Soil Description[2]:	SILT and CLAY - some sand	Sampled By:	Client
		USC Classification:	Cu: #N/A
Moisture Content:	35.9%		Cc: #N/A

Particle Size (mm)	Percent Passing
75	
50	
38	
25	
19	
12.5	
10	
5	100
2	99
0.85	98
0.425	97
0.25	96
0.15	94
0.075	86.8
0.0304	69.5
0.0196	64.6
0.0116	57.7
0.0083	54.8
0.0060	47.9
0.0030	33.3
0.0014	17.6



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

Reviewed By: _____ C.E.T.

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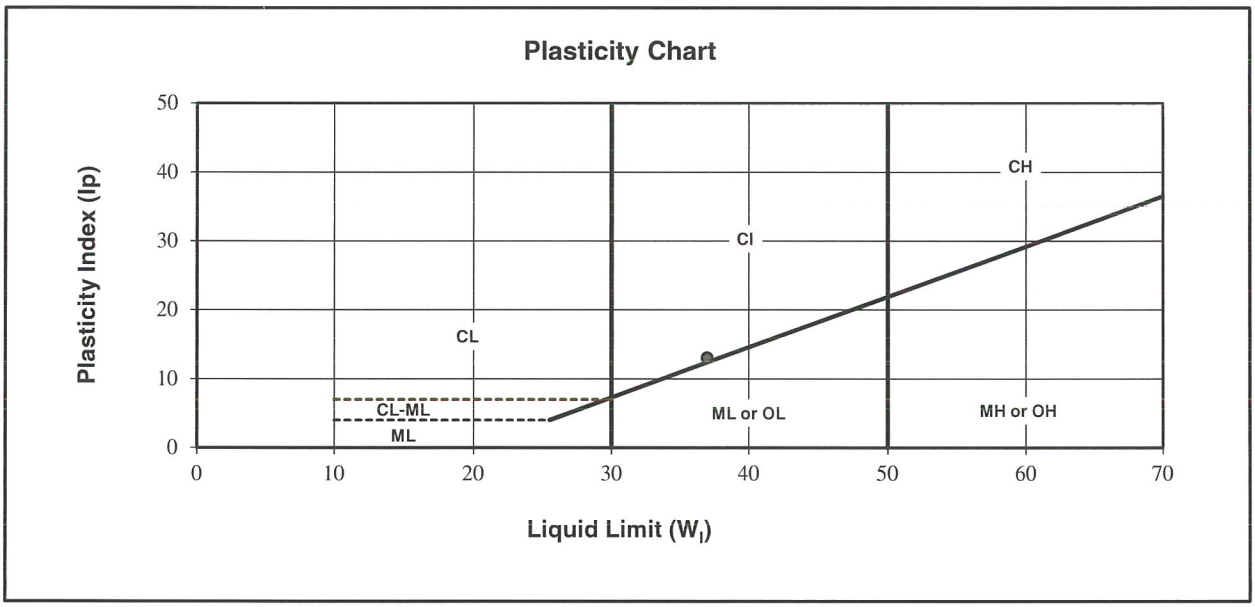
ATTERBERG LIMITS TEST REPORT

ASTM D4318

Project:	Minto February 2015 Lab Testing	Sample Number:	MM-101285
		Borehole Number:	14-SWC-973
Project No:	W14103546-01	Depth:	2.80 - 3.10 m
Client:	SRK Consulting (Canada) Inc.	Sampled By:	Client
Attention:	Murrary McGregor	Date Sampled:	November 6, 2014
Email:		Date Tested:	March 5, 2015

Tested By: AMT

Sample Description: SILT and CLAY - some sand



Liquid Limit ($W_{1)}$:	37	Natural Moisture (%)	35.9
Plastic Limit :	24	Soil Plasticity:	Medium
Plasticity Index (Ip) :	**13**	Mod.USCS Symbol:	CI

Remarks:

Reviewed By: C.E.T.

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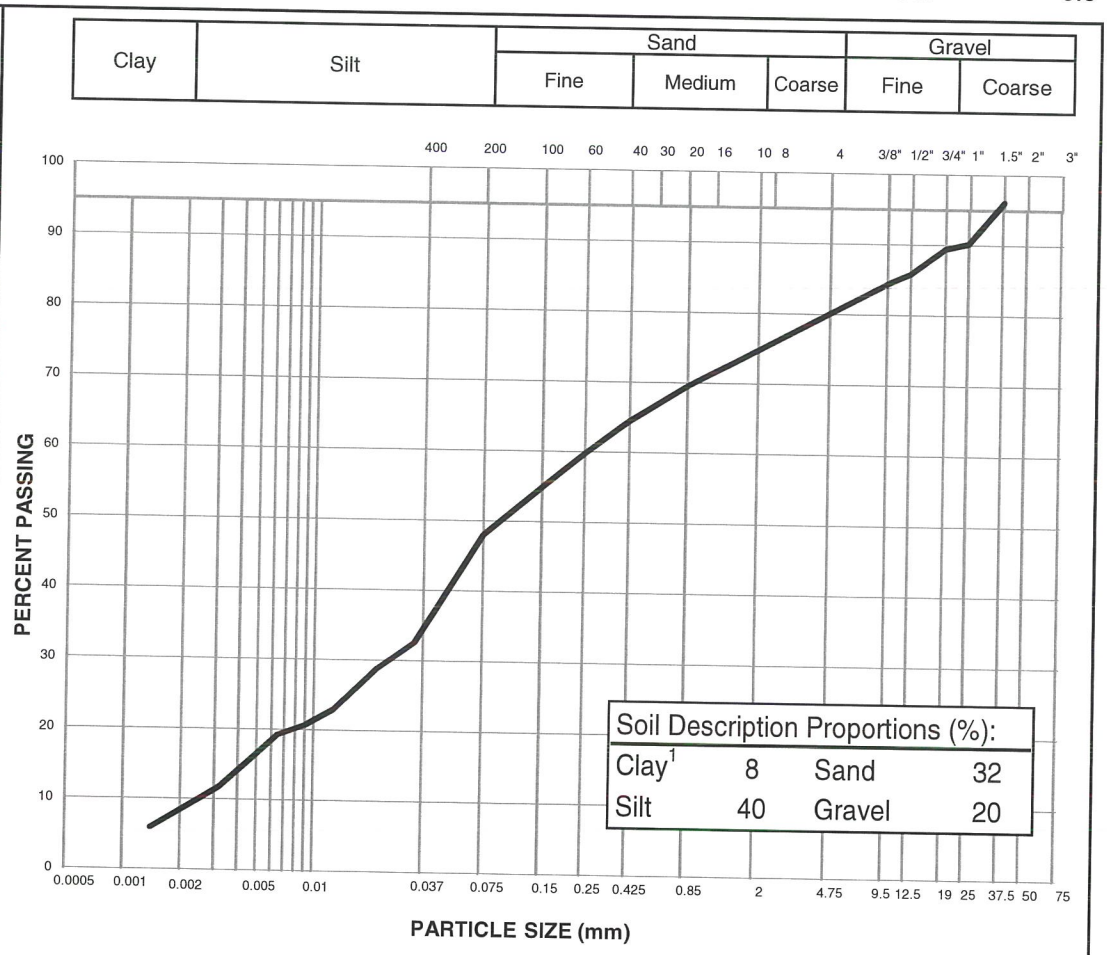
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	MM-101288
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	14-SWC-973
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	9.17 - 9.47 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 3, 2015 By: AMT	Date sampled:	November 6, 2014
Soil Description[2]:	SILT - sandy, some gravel, trace clay	Sampled By:	Client
		USC Classification:	Cu: 97.2
Moisture Content:	13.2%		Cc: 0.9

Particle Size (mm)	Percent Passing
75	
50	
38	96
25	90
19	89
12.5	86
10	85
5	80
2	75
0.85	70
0.425	64
0.25	60
0.15	55
0.075	48.0
0.0335	32.6
0.0215	28.9
0.0127	22.9
0.0091	20.7
0.0065	19.2
0.0032	11.8
0.0014	5.9



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

Reviewed By: _____ C.E.T.

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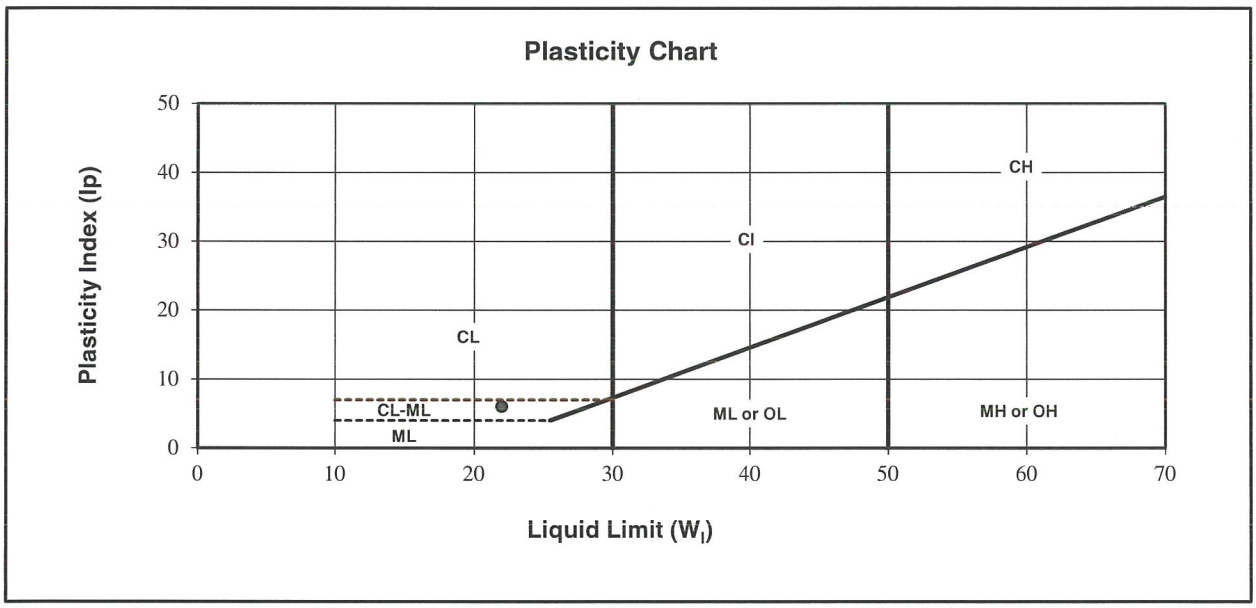
TETRA TECH

ATTERBERG LIMITS TEST REPORT

ASTM D4318

Project:	Minto February 2015 Lab Testing	Sample Number:	MM-101288
		Borehole Number:	14-SWC-973
Project No:	W14103546-01	Depth:	9.17 - 9.47 m
Client:	SRK Consulting (Canada) Inc.	Sampled By:	Client
		Tested By:	AMT
Attention:	Murrary McGregor	Date Sampled:	November 6, 2014
Email:		Date Tested:	March 5, 2015

Sample Description: SILT - sandy, some gravel, trace silt



Liquid Limit (W_1) :	22	Natural Moisture (%)	35.9
Plastic Limit :	16	Soil Plasticity:	Low
Plasticity Index (Ip) :	**6**	Mod.USCS Symbol:	CL-ML

Remarks:

Reviewed By: _____ C.E.T.

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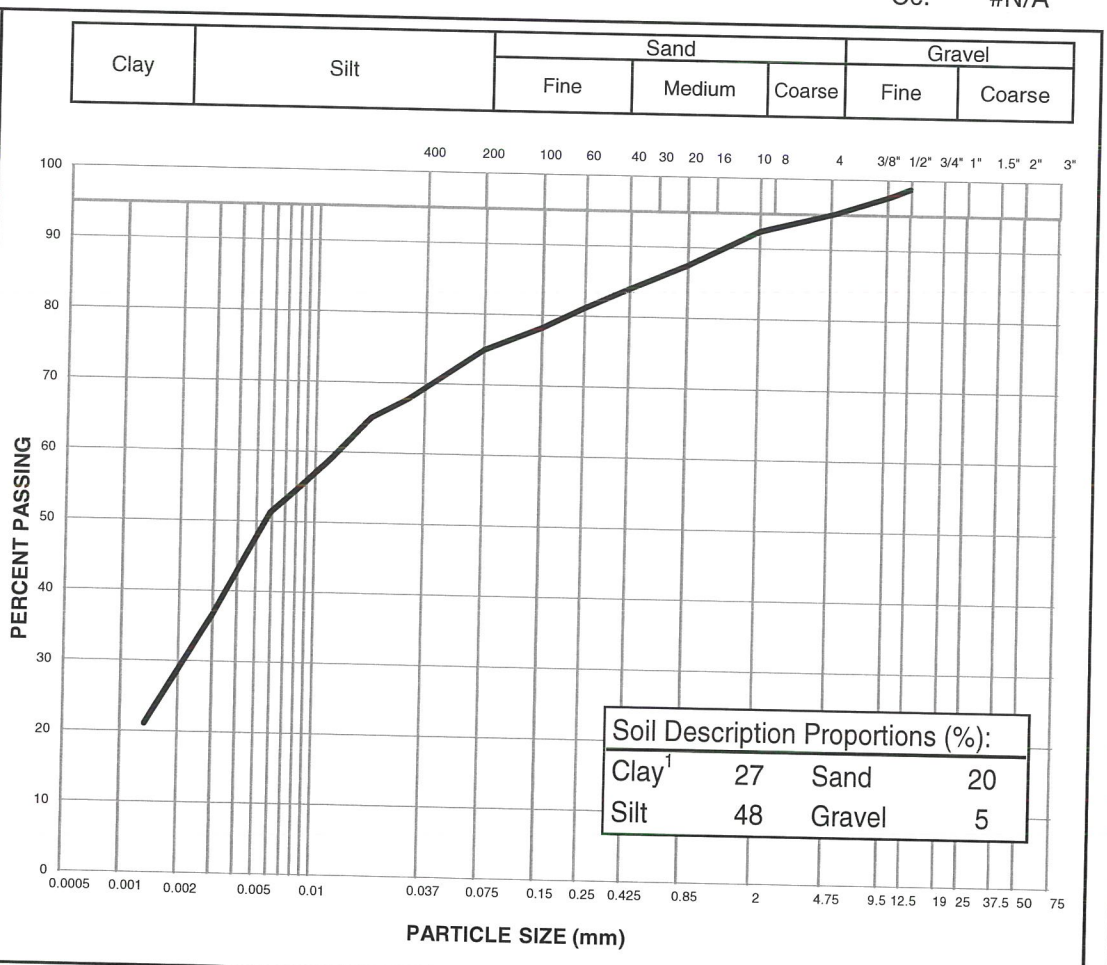
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	MM-101298
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	14-SWC-975
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	5.70 - 6.00 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 3, 2015 By: AMT	Date sampled:	November 9, 2014
Soil Description[2]:	SILT and CLAY - some sand, trace gravel	Sampled By:	Client
		USC Classification:	Cu: #N/A
Moisture Content:	28.1%		Cc: #N/A

Particle Size (mm)	Percent Passing
75	
50	
38	
25	
19	
12.5	99
10	98
5	95
2	92
0.85	87
0.425	84
0.25	81
0.15	78
0.075	74.9
0.0304	67.8
0.0195	65.0
0.0116	58.6
0.0083	54.9
0.0059	51.3
0.0030	36.6
0.0013	21.1



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification:

Remarks:

Reviewed By: C.E.T.

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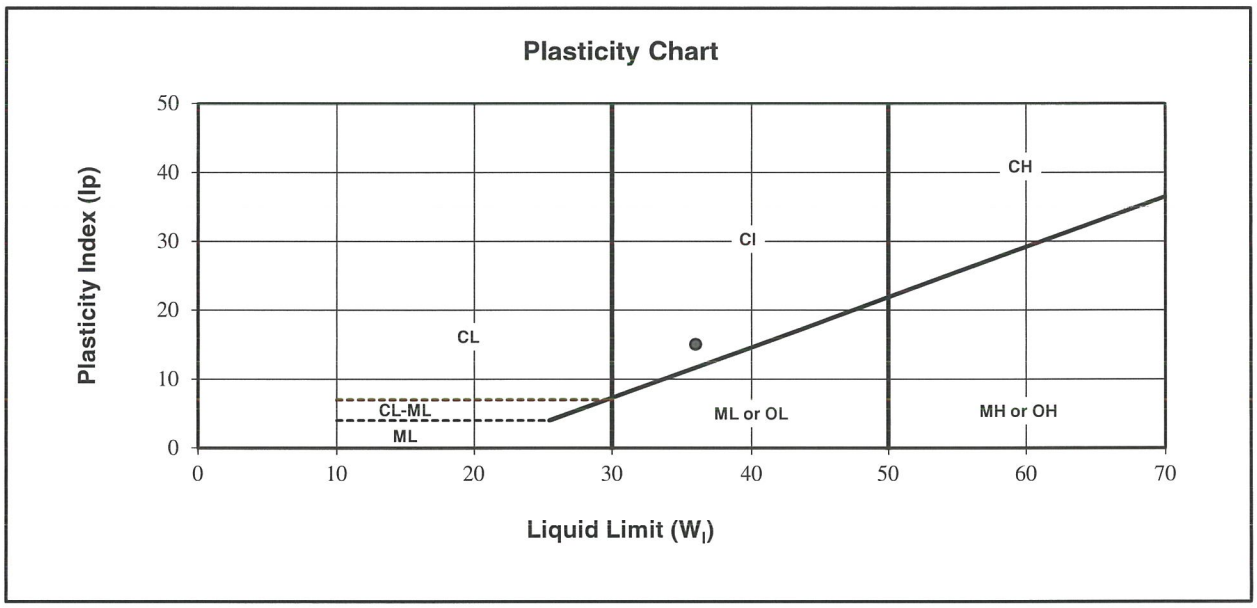
TETRA TECH

ATTERBERG LIMITS TEST REPORT

ASTM D4318

Project:	Minto February 2015 Lab Testing	Sample Number:	MM-101298
		Borehole Number:	14-SWC-975
Project No:	W14103546-01	Depth:	5.70 - 6.00 m
Client:	SRK Consulting (Canada) Inc.	Sampled By:	Client Tested By: AMT
Attention:	Murrary McGregor	Date Sampled:	November 9, 2014
Email:		Date Tested:	March 5, 2015

Sample Description: SILT and CLAY - some sand, trace gravel



Liquid Limit ($W_{1)}$:	36	Natural Moisture (%)	35.9
Plastic Limit :	21	Soil Plasticity:	Medium
Plasticity Index (Ip) :	**15**	Mod.USCS Symbol:	CI

Remarks:

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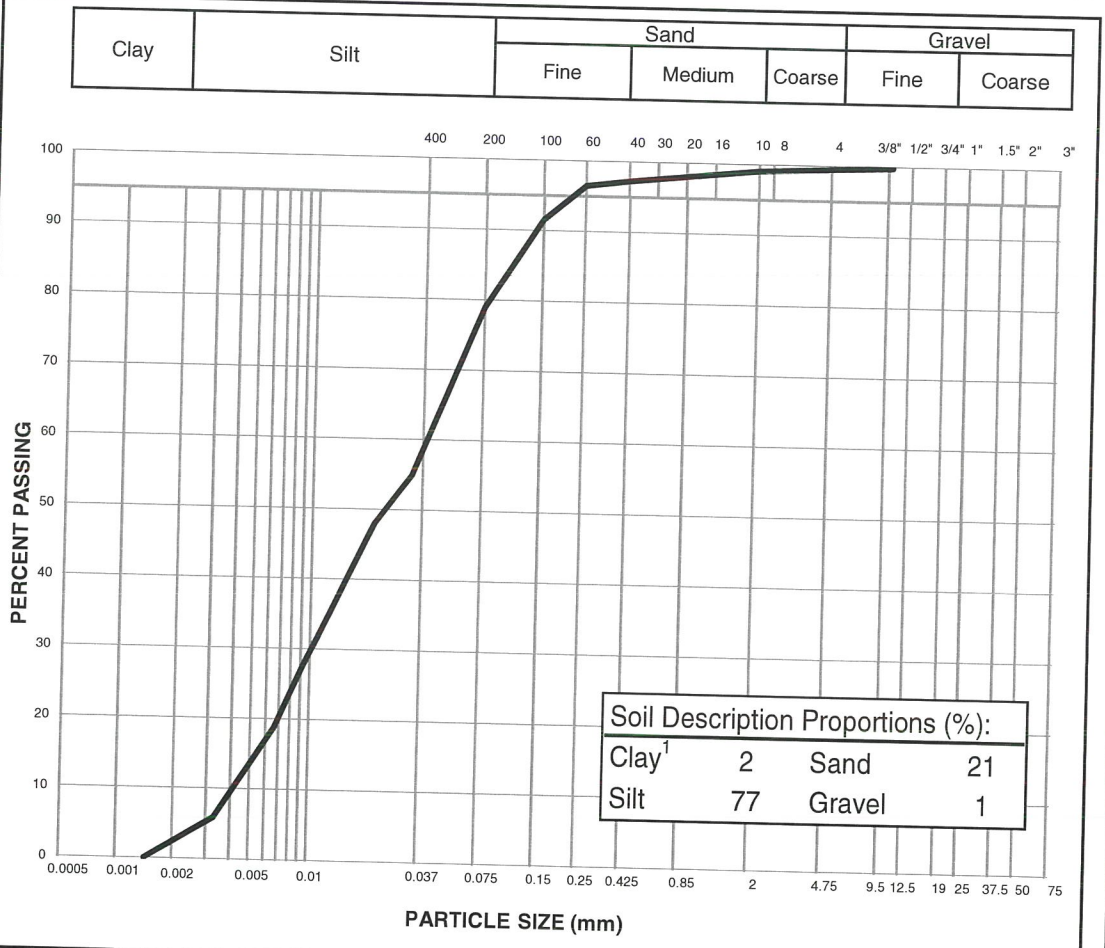
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	MM-101317
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	14-SWC-979
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	20.60 - 20.90 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 3, 2015 By: AMT	Date sampled:	November 9, 2014
Soil Description[2]:	SILT - sandy, trace clay, trace gravel	Sampled By:	Client
		USC Classification:	Cu: 9.5
Moisture Content:	26.5%		Cc: 0.6

Particle Size (mm)	Percent Passing
75	
50	
38	
25	
19	
12.5	
10	100
5	100
2	99
0.85	98
0.425	97
0.25	96
0.15	92
0.075	78.9
0.0325	54.9
0.0211	48.1
0.0127	35.3
0.0092	27.5
0.0066	18.6
0.0033	5.9
0.0014	0.0



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification:

Remarks:

Reviewed By: C.E.T.

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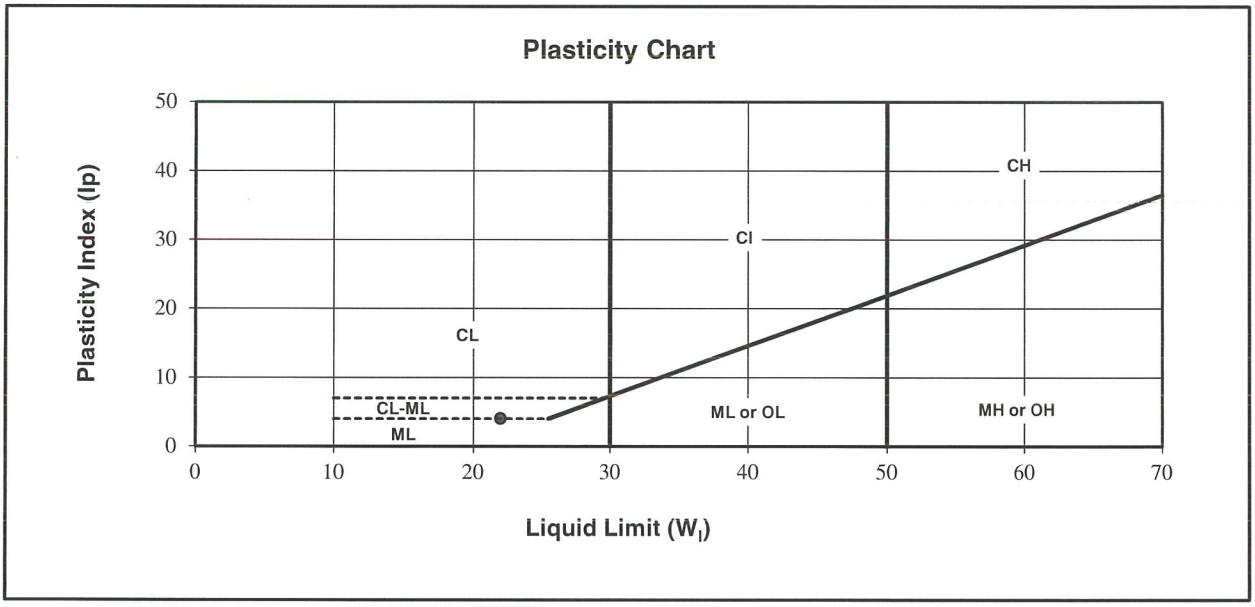
TETRA TECH

ATTERBERG LIMITS TEST REPORT

ASTM D4318

Project:	Minto February 2015 Lab Testing	Sample Number:	MM-101317
		Borehole Number:	14-SWC-979
Project No:	W14103546-01	Depth:	20.60 - 20.90 m
Client:	SRK Consulting (Canada) Inc.	Sampled By:	Client Tested By: AMT
Attention:	Murrary McGregor	Date Sampled:	November 9, 2014
Email:		Date Tested:	March 5, 2015

Sample Description: SILT - sandy, trace clay, trace gravel



Liquid Limit (W_1) :	22	Natural Moisture (%)	35.9
Plastic Limit :	18	Soil Plasticity:	Low
Plasticity Index (Ip) :	4	Mod.USCS Symbol:	CL-ML

Remarks:

Reviewed By: _____ C.E.T.

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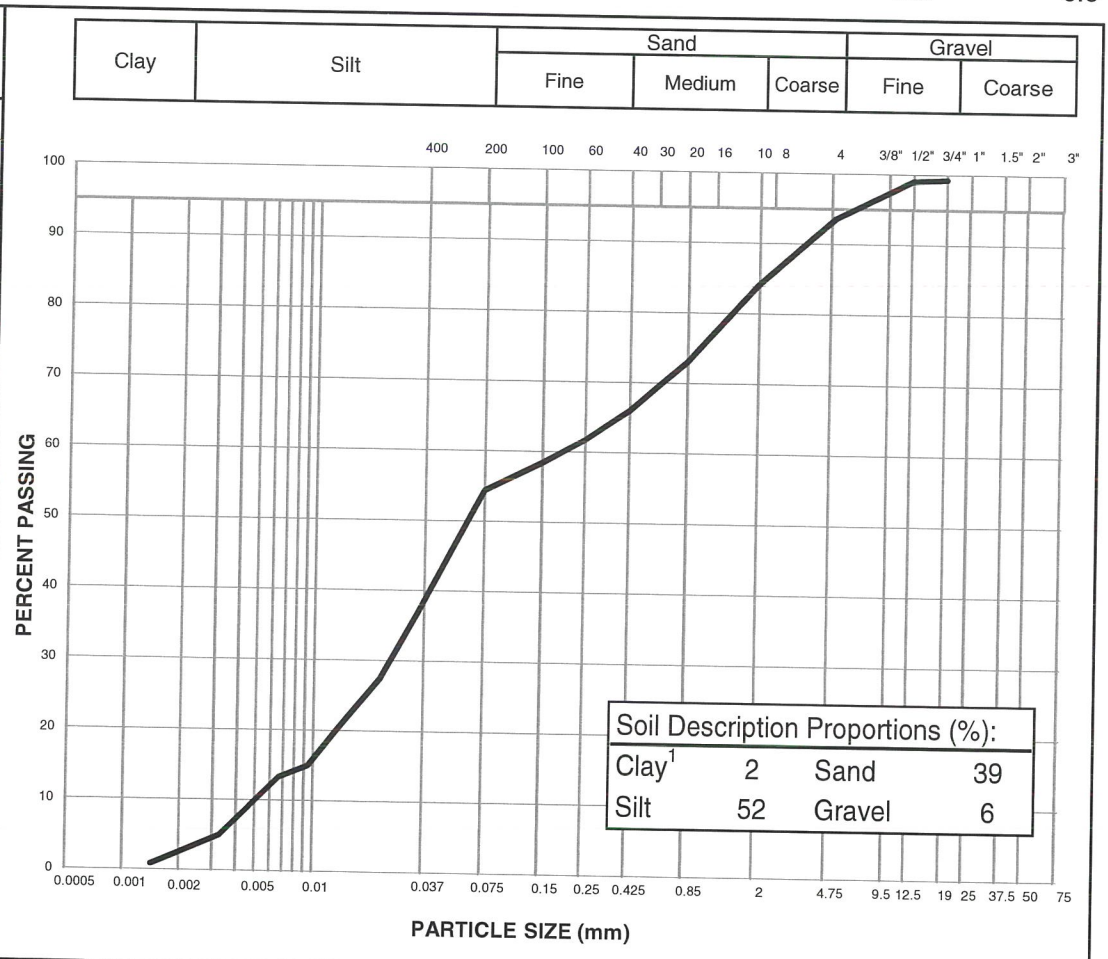
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	MM-101326
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	14-SWC-980
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	19.40 - 19.70 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 3, 2015 By: AMT	Date sampled:	November 9, 2014
Soil Description[2]:	SILT and SAND - trace gravel, trace clay	Sampled By:	Client
		USC Classification:	Cu: 37.1
Moisture Content:	18.8%		Cc: 0.6

Particle Size (mm)	Percent Passing
75	
50	
38	
25	
19	99
12.5	99
10	98
5	94
2	84
0.85	73
0.425	66
0.25	62
0.15	58
0.075	54.4
0.0339	36.7
0.0222	27.5
0.0131	20.0
0.0094	15.0
0.0066	13.3
0.0033	5.0
0.0014	0.8



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

Reviewed By: _____ C.E.T.

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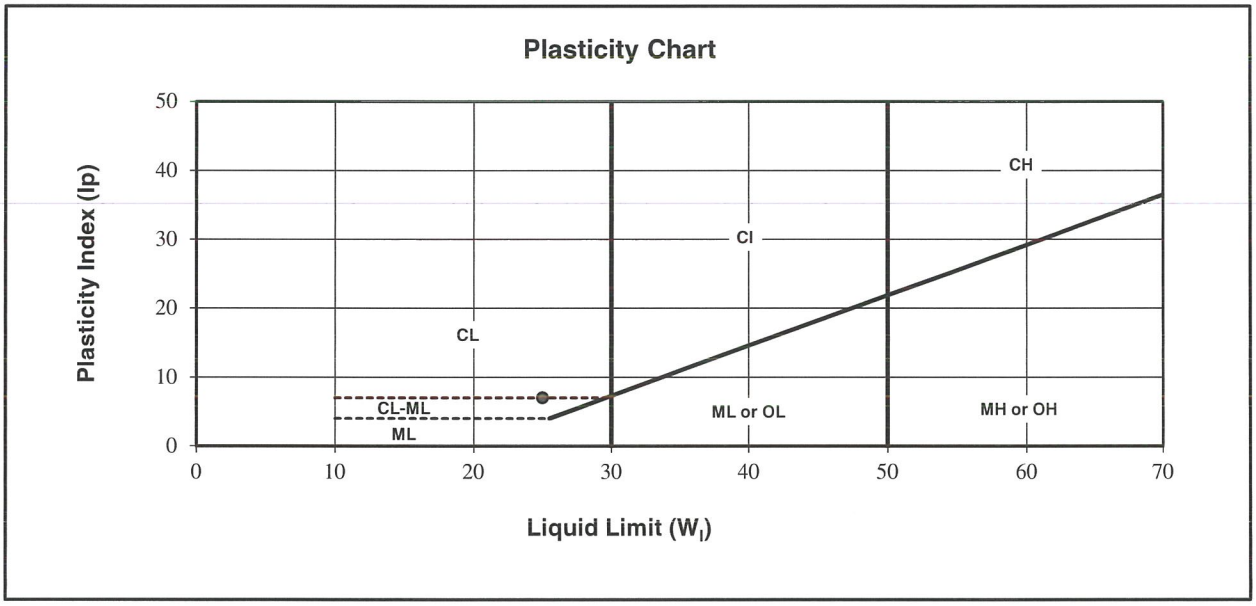
TETRA TECH

ATTERBERG LIMITS TEST REPORT

ASTM D4318

Project:	Minto February 2015 Lab Testing	Sample Number:	MM-101326
		Borehole Number:	14-SWC-980
Project No:	W14103546-01	Depth:	19.40 - 19.70 m
Client:	SRK Consulting (Canada) Inc.	Sampled By:	Client
		Tested By:	AMT
Attention:	Murrary McGregor	Date Sampled:	November 9, 2014
Email:		Date Tested:	March 5, 2015

Sample Description: SILT and SAND - trace gravel, trace clay



Liquid Limit ($W_{1)}$:	25	Natural Moisture (%)	35.9
Plastic Limit :	18	Soil Plasticity:	Low
Plasticity Index (Ip) :	7	Mod.USCS Symbol:	CL-ML

Remarks:

Reviewed By: C.E.T.

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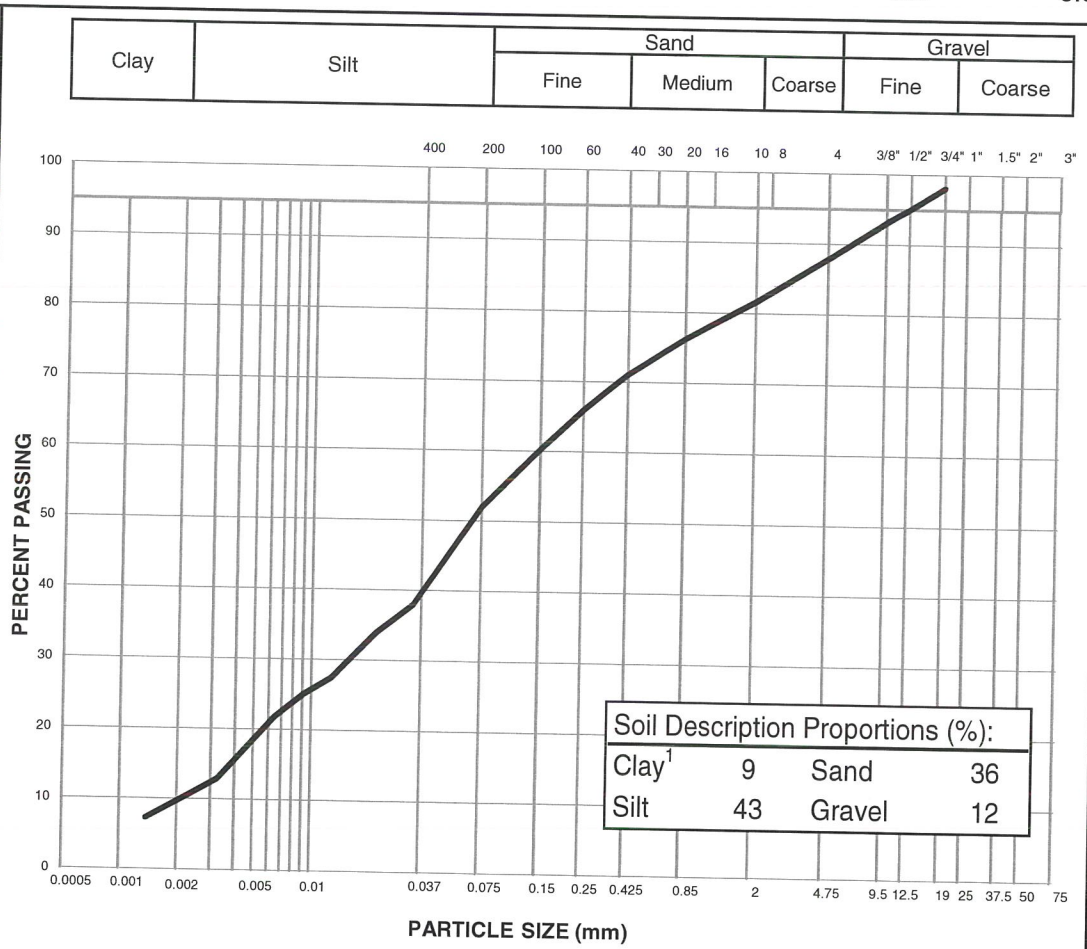
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	58554
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	15-SWC-995
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	8.60 - 8.90 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 4, 2015 By: AMT	Date sampled:	February 9, 2015
Soil Description[2]:	SILT and SAND - some gravel, trace clay	Sampled By:	Client
		USC Classification:	Cu: 64.8
Moisture Content:	13.4%		Cc: 0.8

Particle Size (mm)	Percent Passing
75	
50	
38	
25	
19	98
12.5	95
10	94
5	88
2	82
0.85	76
0.425	71
0.25	66
0.15	60
0.075	51.9
0.0335	38.0
0.0216	33.9
0.0128	27.5
0.0091	25.0
0.0064	21.8
0.0033	12.9
0.0014	7.3



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

Reviewed By: _____ C.E.T.

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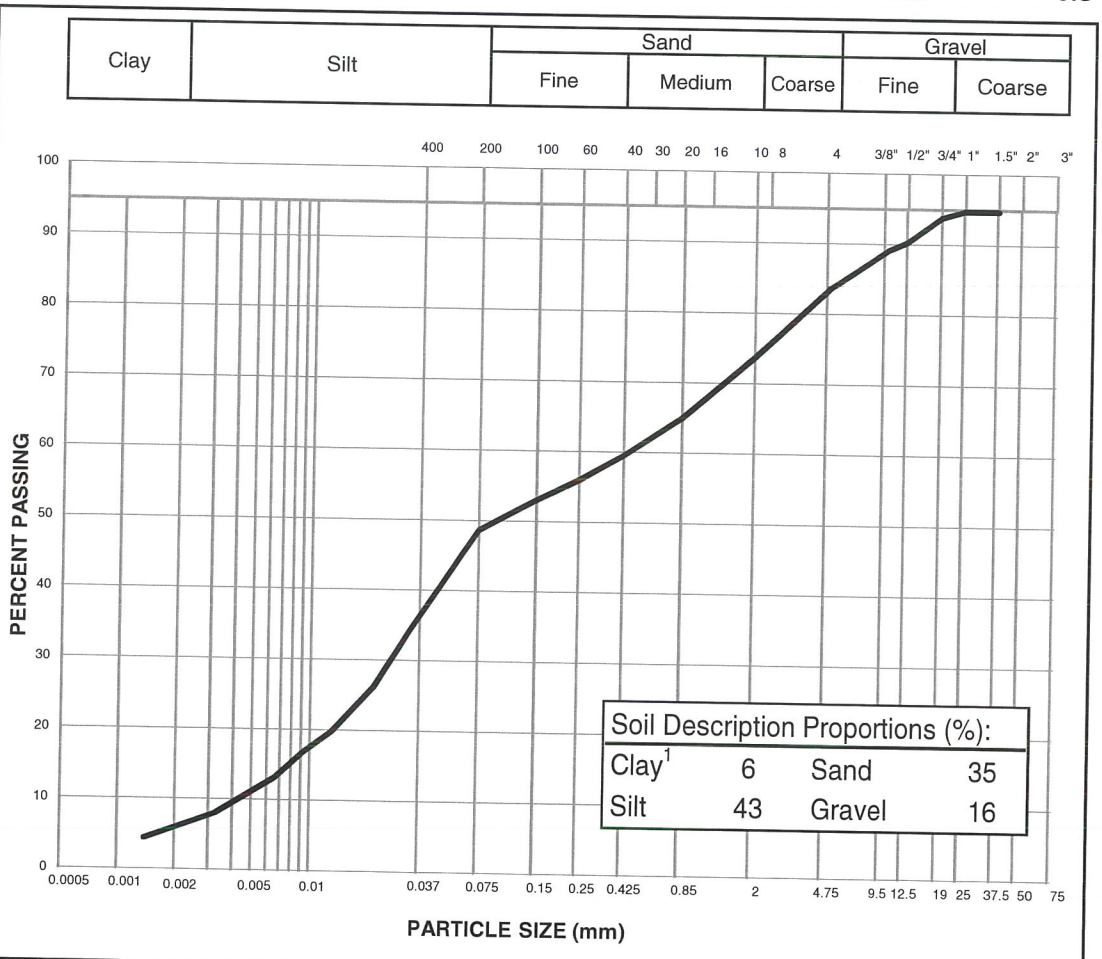
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	58557
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	15-SWC-996
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	6.12 - 6.42 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 4, 2015 By: AMT	Date sampled:	February 9, 2015
Soil Description[2]:	SILT - sandy, some gravel, trace clay	Sampled By:	Client
		USC Classification:	Cu: 101.8
Moisture Content:	14.8%		Cc: 0.3

Particle Size (mm)	Percent Passing
75	
50	
38	95
25	95
19	94
12.5	90
10	89
5	84
2	74
0.85	65
0.425	60
0.25	56
0.15	53
0.075	48.6
0.0327	34.3
0.0214	26.2
0.0129	19.7
0.0092	16.8
0.0066	13.1
0.0033	8.0
0.0014	4.4



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

Reviewed By: _____ C.E.T.

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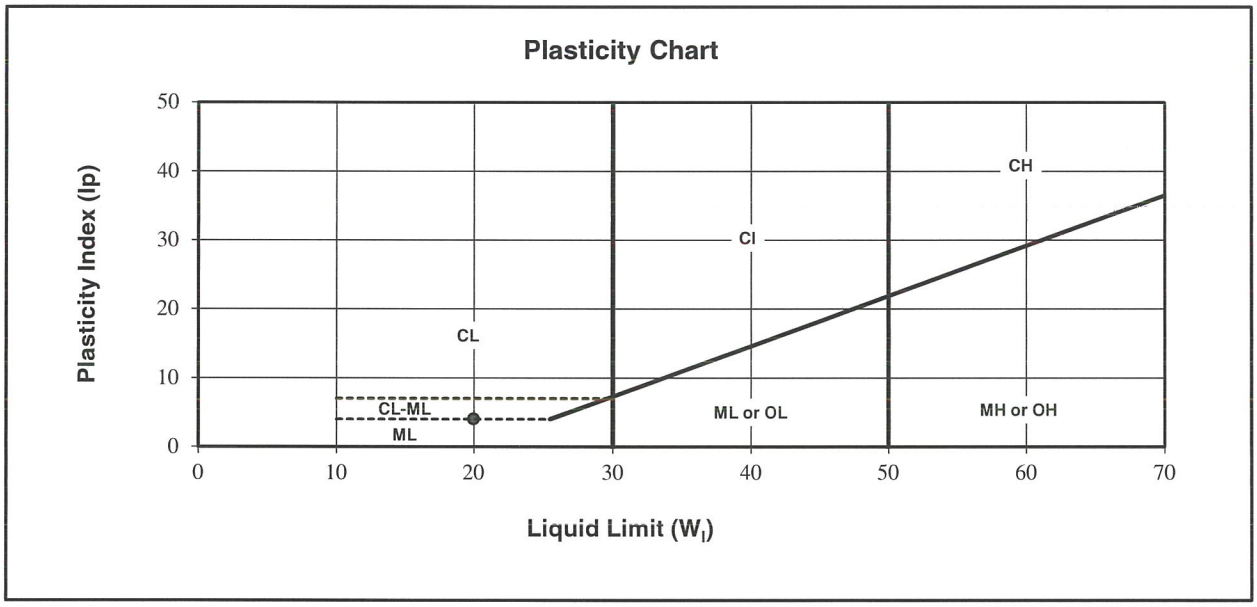
TETRA TECH

ATTERBERG LIMITS TEST REPORT

ASTM D4318

Project:	Minto February 2015 Lab Testing	Sample Number:	58557
		Borehole Number:	15-SWC-996
Project No:	W14103546-01	Depth:	6.12 - 6.42 m
Client:	SRK Consulting (Canada) Inc.	Sampled By:	Client
		Tested By:	AMT
Attention:	Murrary McGregor	Date Sampled:	February 9, 2015
Email:		Date Tested:	March 6, 2015

Sample Description: SILT - sandy, some gravel, trace clay



Liquid Limit ($W_{1)}$:	20	Natural Moisture (%)	35.9
Plastic Limit :	16	Soil Plasticity:	Low
Plasticity Index (Ip) :	4	Mod.USCS Symbol:	CL-ML

Remarks:

Reviewed By: C.E.T.

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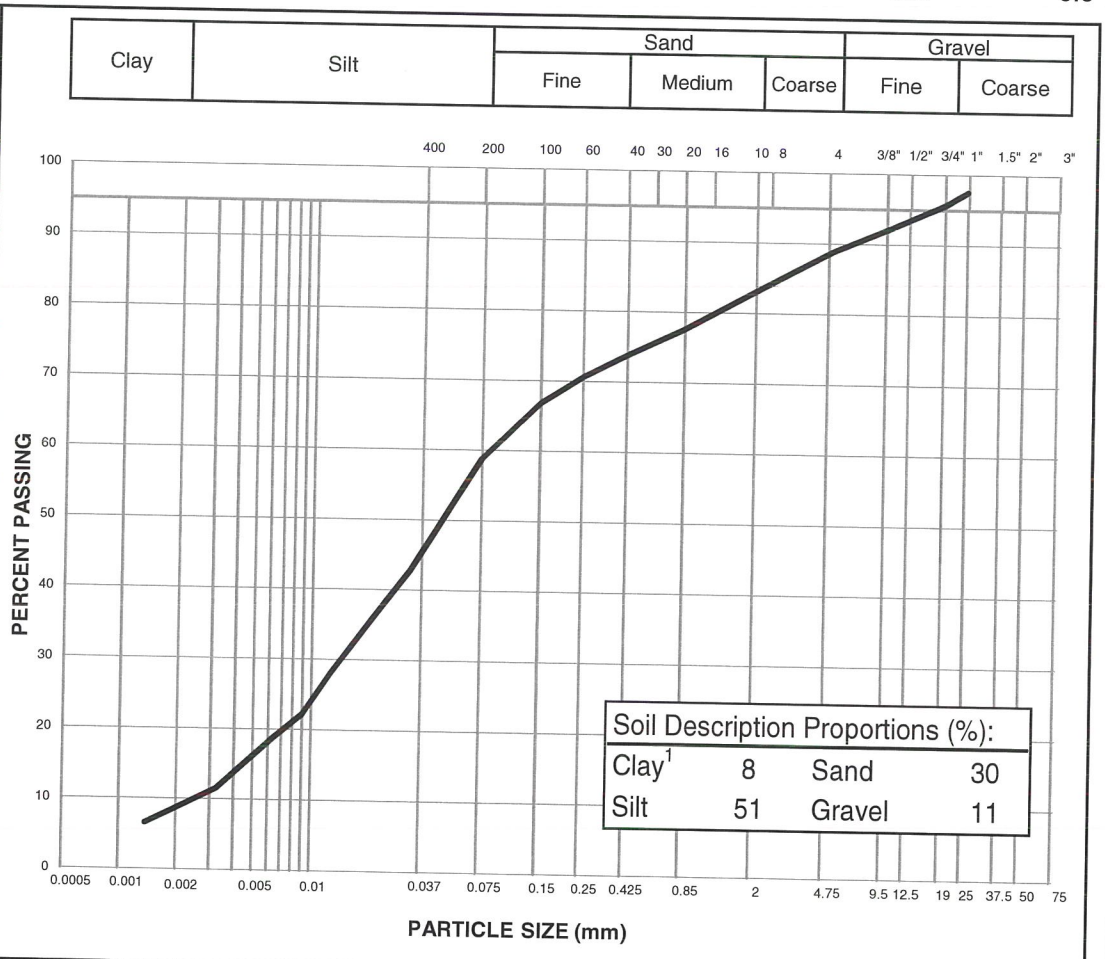
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	58559
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	15-SWC-996
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	14.73 - 15.10 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 5, 2015 By: AMT	Date sampled:	February 9, 2015
Soil Description[2]:	SILT - sandy, some gravel, trace clay	Sampled By:	Client
		USC Classification:	Cu: 32.3
Moisture Content:	16.1%		Cc: 0.9

Particle Size (mm)	Percent Passing
75	
50	
38	
25	98
19	96
12.5	94
10	92
5	89
2	83
0.85	78
0.425	74
0.25	71
0.15	67
0.075	58.8
0.0321	42.8
0.0209	36.2
0.0124	28.0
0.0090	22.2
0.0064	18.9
0.0032	11.5
0.0014	6.6



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

Reviewed By: _____ C.E.T.

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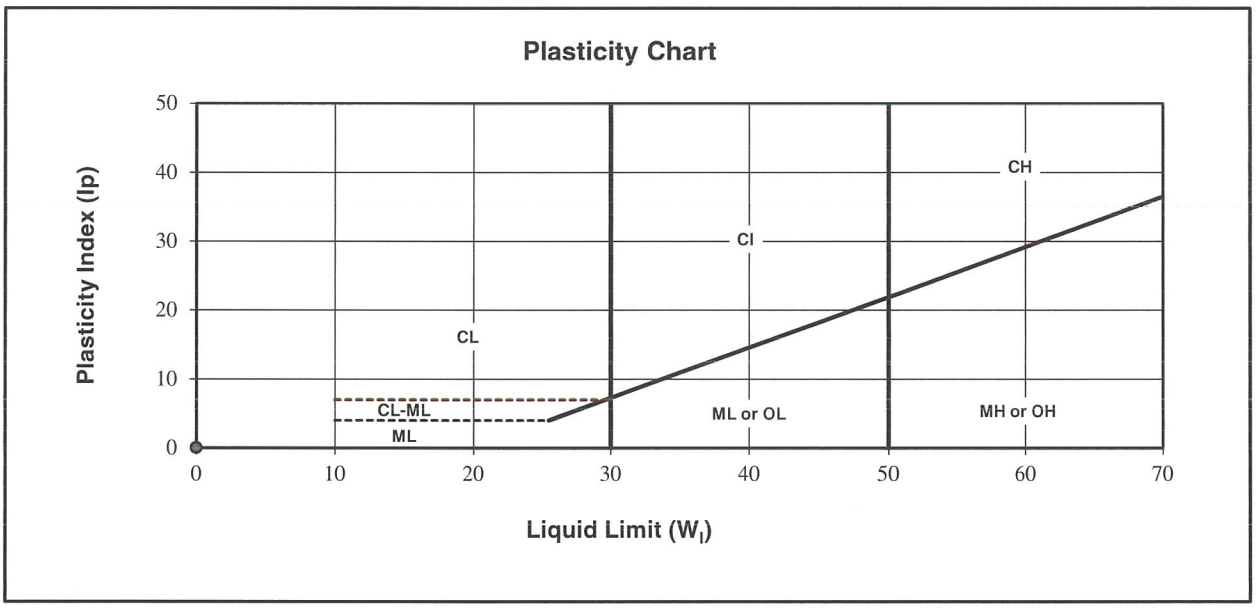
TETRA TECH

ATTERBERG LIMITS TEST REPORT

ASTM D4318

Project:	Minto February 2015 Lab Testing	Sample Number:	58559
		Borehole Number:	15-SWC-996
Project No:	W14103546-01	Depth:	14.73 - 15.10 m
Client:	SRK Consulting (Canada) Inc.	Sampled By:	Client Tested By: AMT
Attention:	Murrary McGregor	Date Sampled:	February 9, 2015
Email:		Date Tested:	March 6, 2015

Sample Description: SILT - sandy, some gravel, trace clay



Liquid Limit ($W_{1)}$:	0	Natural Moisture (%)	35.9
Plastic Limit :	0	Soil Plasticity:	NP
Plasticity Index (Ip) :	**0**	Mod.USCS Symbol:	N/A

Remarks: An atterberg was attempted and material was found to be non-plastic.

Reviewed By: _____ C.E.T.

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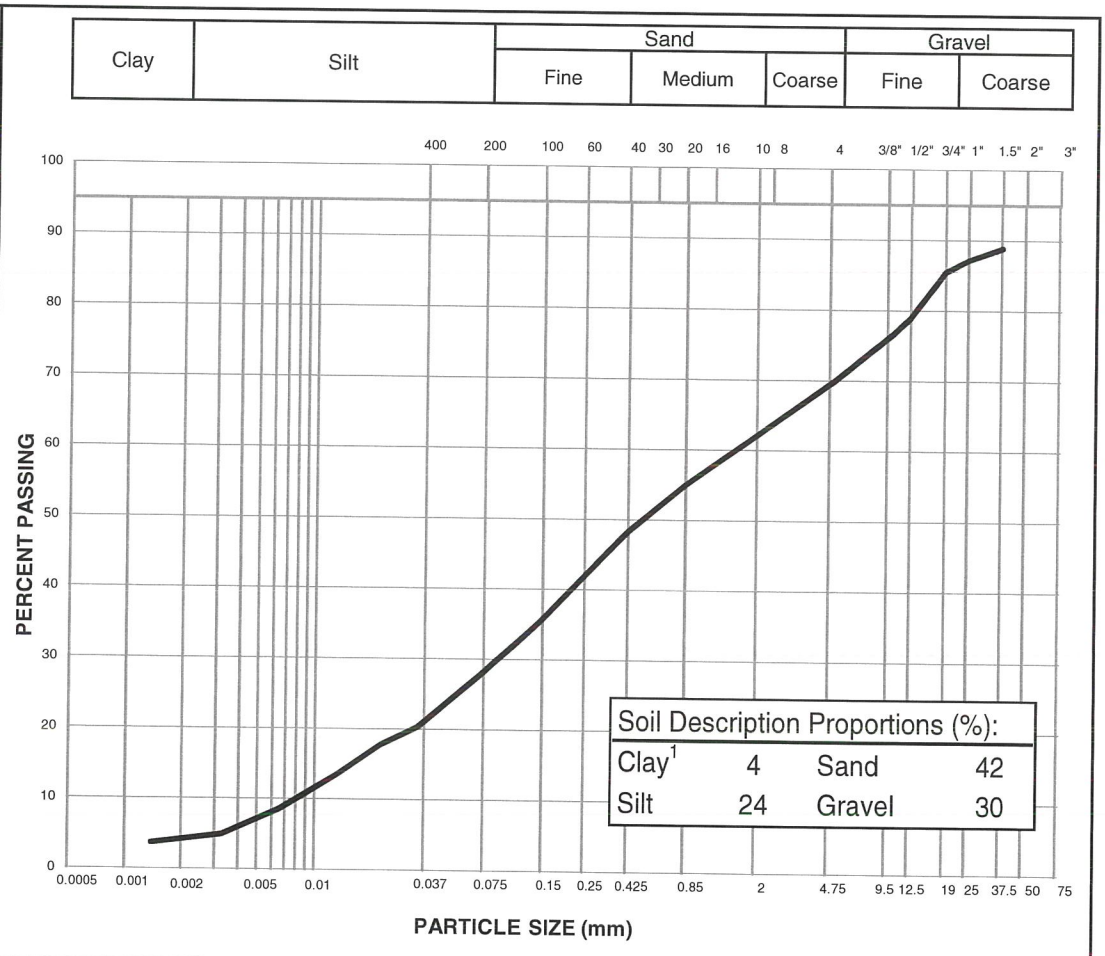
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	160025
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	15-SWC-997
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	9.79 - 10.11 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 5, 2015 By: AMT	Date sampled:	February 9, 2015
Soil Description[2]:	SAND - gravelley, silty, trace clay	Sampled By:	Client
		USC Classification:	Cu: 203.9
Moisture Content:	10.3%		Cc: 0.6

Particle Size (mm)	Percent Passing
75	
50	
38	89
25	87
19	85
12.5	79
10	76
5	70
2	62
0.85	55
0.425	48
0.25	42
0.15	36
0.075	28.2
0.0342	20.3
0.0221	17.8
0.0131	13.5
0.0093	11.1
0.0066	8.6
0.0033	4.9
0.0014	3.7



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

_____

Reviewed By: _____ C.E.T.

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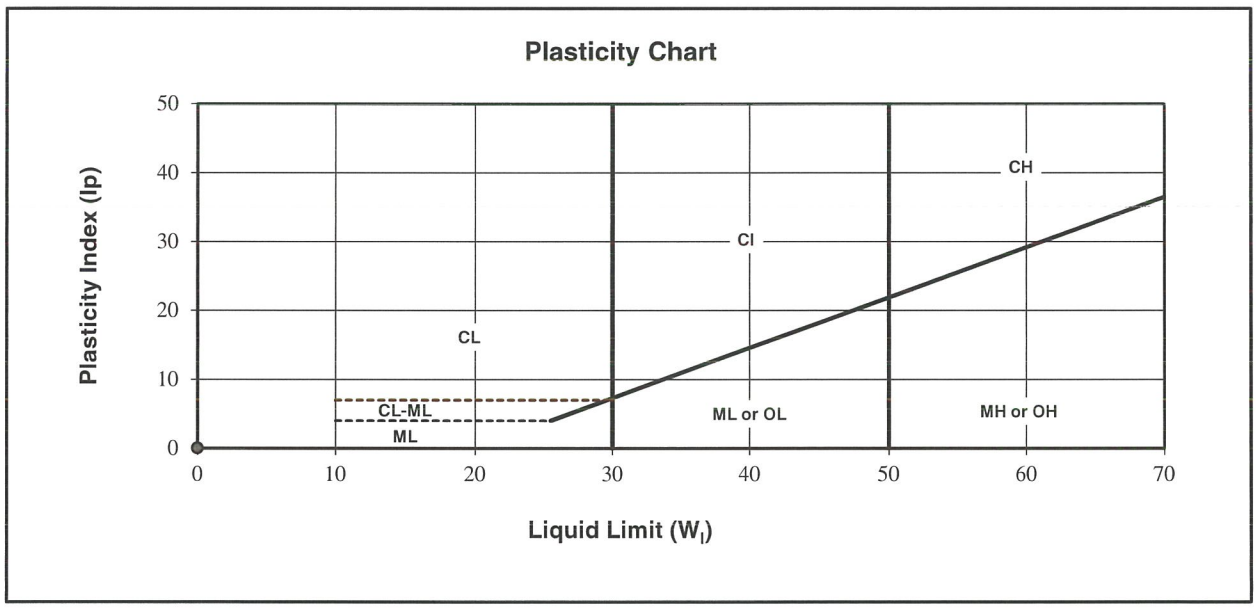
TETRA TECH

ATTERBERG LIMITS TEST REPORT

ASTM D4318

Project:	Minto February 2015 Lab Testing	Sample Number:	160025
		Borehole Number:	15-SWC-997
Project No:	W14103546-01	Depth:	9.79 - 10.11 m
Client:	SRK Consulting (Canada) Inc.	Sampled By:	Client Tested By: AMT
Attention:	Murrary McGregor	Date Sampled:	February 9, 2015
Email:		Date Tested:	March 5, 2015

Sample Description: SAND - gravelley, silty, trace clay



Liquid Limit ($W_{1)}$:	0	Natural Moisture (%)	35.9
Plastic Limit :	0	Soil Plasticity:	NP
Plasticity Index (Ip) :	**0**	Mod.USCS Symbol:	N/A

Remarks: An atterberg was attempted and material was found to be non-plastic.

Reviewed By: _____ C.E.T.

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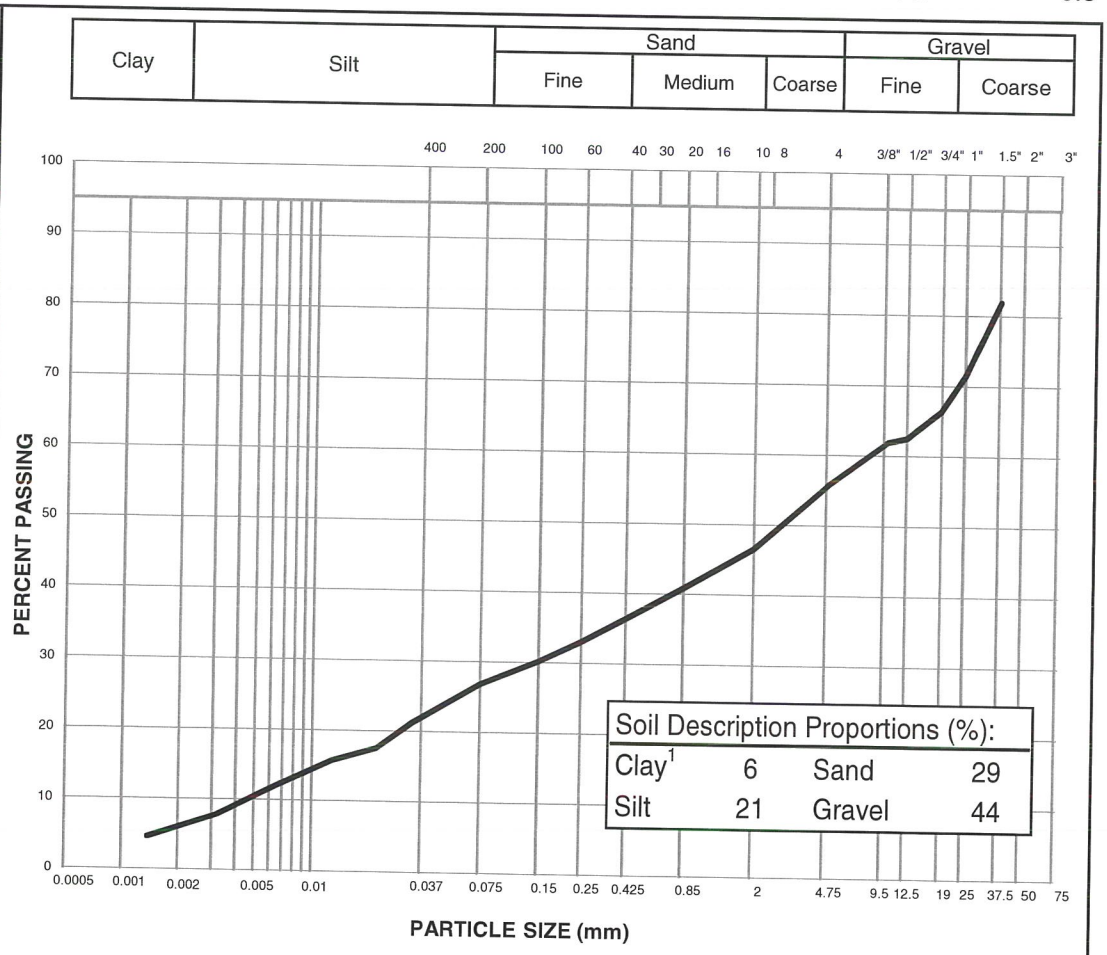
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	58568
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	15-SWC-997
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	23.09 - 23.43 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 5, 2015 By: AMT	Date sampled:	February 9, 2015
Soil Description[2]:	GRAVEL - sandy, silty, trace clay	Sampled By:	Client
		USC Classification:	Cu: 1712.3
Moisture Content:	11.7%		Cc: 0.5

Particle Size (mm)	Percent Passing
75	
50	
38	82
25	71
19	66
12.5	63
10	62
5	56
2	47
0.85	41
0.425	36
0.25	33
0.15	30
0.075	26.8
0.0332	21.2
0.0216	17.5
0.0126	15.7
0.0090	13.8
0.0065	12.0
0.0032	7.8
0.0014	4.6



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

Reviewed By: _____ C.E.T.

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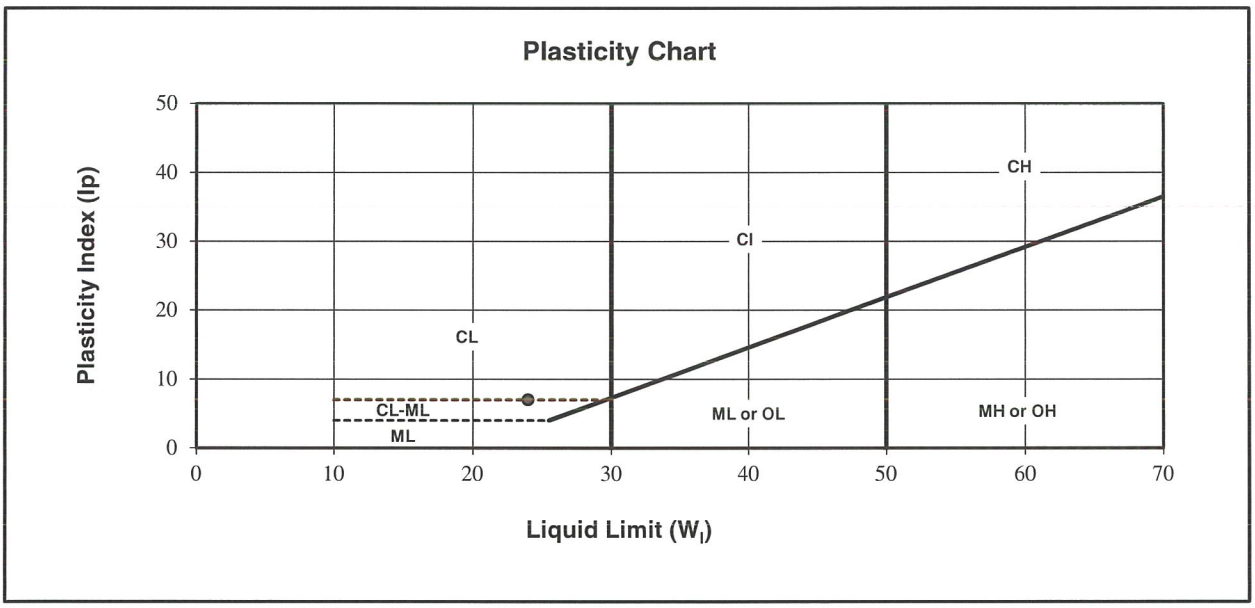
TETRA TECH

ATTERBERG LIMITS TEST REPORT

ASTM D4318

Project:	Minto February 2015 Lab Testing	Sample Number:	58568
		Borehole Number:	15-SWC-997
Project No:	W14103546-01	Depth:	23.09 - 23.43 m
Client:	SRK Consulting (Canada) Inc.	Sampled By:	Client Tested By: AMT
Attention:	Murrary McGregor	Date Sampled:	February 9, 2015
Email:		Date Tested:	March 5, 2015

Sample Description: GRAVEL - sandy, silty, trace clay



Liquid Limit ($W_{1)}$:	24	Natural Moisture (%)	35.9
Plastic Limit :	17	Soil Plasticity:	Low
Plasticity Index (Ip) :	7	Mod.USCS Symbol:	CL-ML

Remarks:

Reviewed By: C.E.T.

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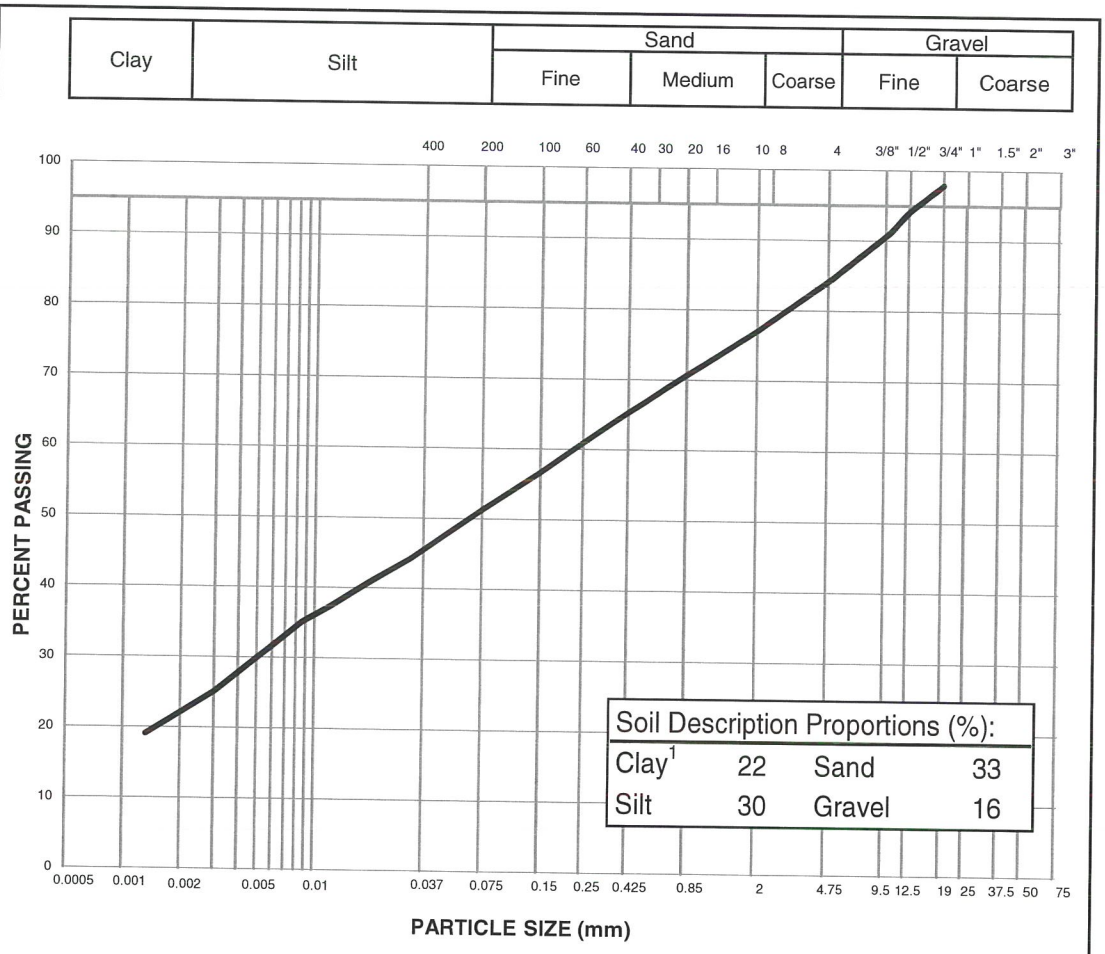
TETRA TECH

PARTICLE SIZE ANALYSIS REPORT

ASTM D422, C136 & C117

Project:	Minto February 2015 Lab Testing	Sample No.:	58572
Project No.:	W14103546-01	Material Type:	
Site:	Minto Mine, YT	Sample Loc.:	15-SWC-997
Client:	SRK Consulting (Canada) Inc.	Sample Depth:	41.32 - 41.67 m
Client Rep.:	Murray McGregor	Sampling Method:	Grab
Date Tested:	March 5, 2015 By: AMT	Date sampled:	February 9, 2015
Soil Description[2]:	CLAY - sandy, silty, some gravel	Sampled By:	Client
		USC Classification:	Cu: #N/A
Moisture Content:	12.7%		Cc: #N/A

Particle Size (mm)	Percent Passing
75	
50	
38	
25	
19	98
12.5	94
10	91
5	85
2	77
0.85	71
0.425	65
0.25	61
0.15	57
0.075	51.3
0.0315	44.2
0.0202	41.2
0.0120	37.3
0.0086	35.1
#N/A	28.7
0.0030	25.2
0.0013	19.1



Notes: [1] The upper clay size of 2 um, per the Canadian Foundation Engineering Manual

[2] The description is visually based & subject to EBA description protocols

Specification: _____

Remarks: _____

_____

Reviewed By: _____ C.E.T.

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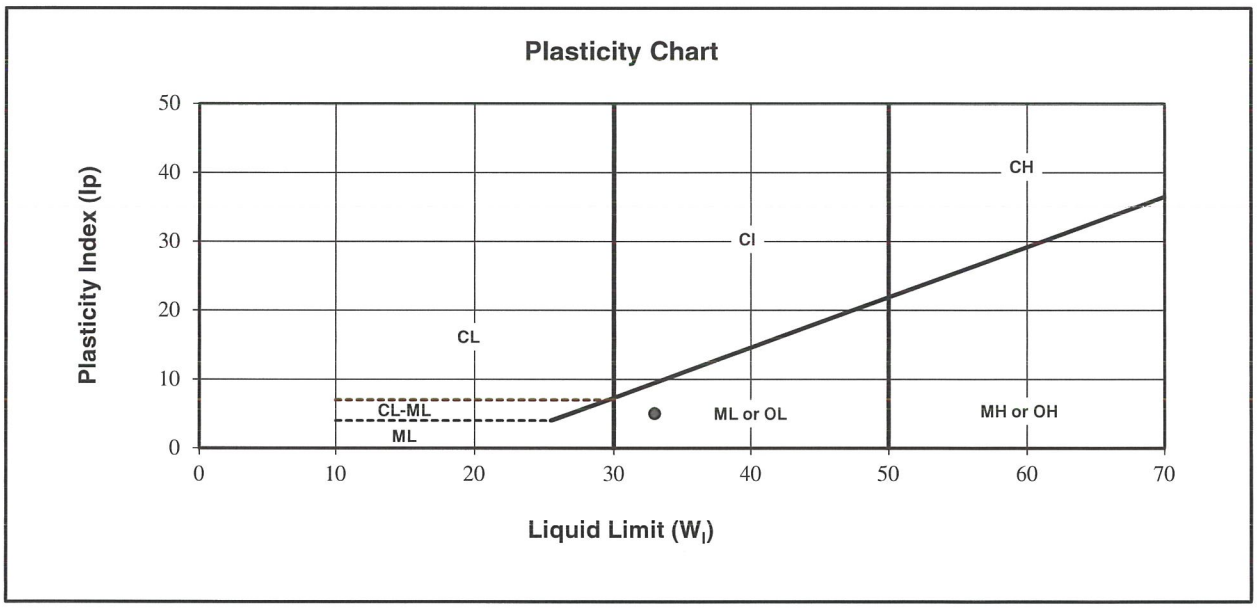
TETRA TECH

ATTERBERG LIMITS TEST REPORT

ASTM D4318

Project:	Minto February 2015 Lab Testing	Sample Number:	58572
		Borehole Number:	15-SWC-997
Project No:	W14103546-01	Depth:	41.32 - 41.67 m
Client:	SRK Consulting (Canada) Inc.	Sampled By:	Client
Attention:	Murrary McGregor	Tested By:	AMT
Email:		Date Sampled:	February 9, 2015
		Date Tested:	March 6, 2015

Sample Description: CLAY - sandy, silty, some gravel



Liquid Limit ($W_{1)}$:	33	Natural Moisture (%)	12.7
Plastic Limit :	28	Soil Plasticity:	Low
Plasticity Index (Ip) :	**5**	Mod.USCS Symbol:	ML

Remarks:

Reviewed By: _____ C.E.T.

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Appendix C – Minto Closure Cover Design – Cover System Erosion Analysis

SRK Consulting (Canada) Inc.
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www.srk.com

Memo

To:	Project File	**Client:**	Minto Exploration Ltd.
From:	Jordan Graham, EIT, Erik Ketilson, PEng.	**Project No:**	1CM002.49
Reviewed by:	Maritz Rykaart, PEng.	**Date:**	July 26, 2016
Subject:	Minto Closure Cover Design - Cover System Erosion Analysis		

1 Introduction

SRK Consulting (Canada) Inc. (SRK) is currently undertaking the update to the closure plan for the Minto Site. Understanding the erosion process is important such that suitable landform designs can be completed, as erosion can significantly alter an engineered landscape.

The purpose of this memo is twofold: first to update the previous erosion analysis completed as part of the closure landform design and reclamation landform unit work completed by SRK (SRK, 2016); and secondly to present the potential effects of erosion due to sheet and rill water erosion that could occur on the engineered slopes at the Site, and evaluate a range of conditions and parameters to help guide the landform designs at Minto. Sheet and rill water erosion occurs as a result of flows that are not concentrated into a particular flow path.

Erosion that may occur within channel flow and the necessary protection to avoid channel erosion is not addressed in this memo.

All calculated erosion estimates are presented as “soil loss”. Soil loss is a mass or depth of eroded material that leaves the slope entirely. Therefore, the estimates within this memo are not representative of the total volume of material that is displaced by water. Although the calculation does not report material that is detached and deposited along the slope, it is factored into the overall calculation.

2 Soil Loss Estimation Methods

There are several methods available for estimating water erosion including the Universal Soil Loss Equation (USLE) (USDA, 1978), the Revised Universal Soil Loss Equation (RUSLE) Versions 1 (USDA, 1997) and 2 (USDA, 2008), the Revised Universal Soil Loss Equation for Use in Canada (RUSLEFAC) (Wall, 2002), the Water Erosion Prediction Project (WEPP) (Flanagan, 2007), SIBERIA (Willgoose, 2005), and many others. Most of these programs take several factors into account to compute soil loss such as climate, topography, soil type, vegetation, and land management practices. The key difference between these methods is that some are based on empirical data while others are based on a mathematical approach using soil physics. The USLE

and its variations are largely based on empirical data, while WEPP and SIBERIA are based on soil physics. RUSLE Version 2 is based on empirical data, but uses soil physics to fill in gaps in empirical data.

The USLE was developed in 1960 and then revised in 1978 (RUSLE) by the United States Department of Agriculture. The empirical relationships in the RUSLE were modified by the Provincial and Federal Governments in 2001 for use in Canada (RUSLEFAC) (Wall et al., 2002). The RUSLEFAC uses metric units and input parameters that apply to Canadian conditions.

The soil loss analysis described within this Memo uses only the RUSLEFAC method. The RUSLEFAC has an advantage over other current methods in that it can be calculated manually and the effects related to the variability of each of the input parameter can be thoroughly evaluated.

3 RUSLEFAC Scope and Limitations

The RUSLEFAC (Wall et al., 2002) is a tool for calculating sheet flow erosion and rill erosion, and as stated in Section 2, is based on empirical data. The experimental soil plots used to develop the equations were subjected to conditions that generally reflected average annual climatic conditions. Therefore, the intent of the RUSLEFAC is to produce a numerical representation of an average annual quantity of soil loss in the units of tonnes per hectare per year, which can be converted to depth per year given an understanding of the soil's in-situ density. The equation is a useful tool for long term predictions, and can also be used for short term losses; however, due to the nature of the experimental data that was collected to develop the equations, short term estimates are likely associated with a greater degree of error.

The RUSLEFAC has the following limitations (Wall et al., 2002):

- It does not accurately estimate soil loss from a single rainfall event. However, the erosivity of a single storm can be estimated using the method described in the RUSLE;
- It does not account for erosional losses once gullies or streams form;
- Although there is some account for erosional losses due to snow melt, the equation does not account for this loss with great accuracy; and
- Freeze/thaw can cause ice lenses in soil that will affect the rate of soil loss: the RUSLEFAC does not take this into account.

Ice lensing is typically a greater issue in areas where repeated freeze/thaw cycles occur during one winter season. At the site, however, the surface material is more likely to freeze in the fall and stay frozen throughout the winter and into the spring without repetitive freeze/thaw action. Therefore, the impact of freeze-thaw on the results of the analysis for the Minto site is not considered to be a major influencing factor on erosion of the cover.

4 Design Criteria

Table 4-1 presents the soil erosion classes included in the RUSLEFAC.

Table 4-1: Soil Erosion Classes

Soil Erosion Class	Potential Soil Loss (T/ha/year)
1. Very Low (i.e. tolerable)	< 6
2. Low	6-11
3. Moderate	11-22
4. High	22-33
5. Severe	> 33

For the Minto site, in an effort to minimize erosion of cover material, there is a preference to achieve a Class 1 soil erosion class. In cases where the native Minto soils may not naturally meet a Class 1 soil erosion classification, additional mitigation measures, or support practices may be necessary to achieve very low rates of erosion. The RUSLEFAC considers Class 1 soils to have:

> "Slight to no erosion potential. Minimal erosion problems should occur if good soil conservation management methods are used... A tolerable soil loss (<6 T/ha/year) is the maximum annual amount of soil which can be removed before the long term natural soil productivity of a hillslope is adversely affected." (Wall et al., 2002).

5 RUSLEFAC Equation

The RUSLEFAC equation is calculated manually by first determining several inputs. The RUSLEFAC equation is:

$$A = RKLSCP$$

Where,

A is the potential long term average annual soil loss in tonnes per hectare. *A* can be converted to depth per year if the density of the soil is known.

R is the rainfall factor, which is expressed in energy multiplied by depth over area times duration (MJmm/hah), is calculated using the equation:

$$R = EI$$

Where *E* is the volume of rainfall and runoff (mm/ha) and *I* is the prolonged peak rate of detachment that occurs with runoff (MJ/h).

- R value contours (isoerodent maps) have been developed by the Government of Canada and are included in the RUSLEFAC document (Wall et al., 2002). To

determine the R value in a particular area, interpolation between contours is often required.

- R can be calculated for a single storm event using the R equation if the storm distribution is known or can be estimated.

K is the soil erodibility factor, which is expressed in terms of area multiplied by duration over energy times depth (hah/MJmm).

- *K* is dependent on the sand content, fine sand content, silt content, organic matter content, soil structure, and permeability of the soil.
- *K* is determined by applying the appropriate parameters to the soil erodibility nomograph included in the RUSLEFAC.

L is the length of slope factor (dimensionless).

S is the slope steepness factor (dimensionless).

- *L and S* are typically presented as a single value.
- The *LS* factor represents a ratio of soil loss in comparison to a “standard plot”, which is an experimental plot that has a steepness of 9% and a slope length of 22.13 m. Charts based on experimental data are included in the RUSLEFAC document (Wall et al., 2002), which is used to determine the *LS* factor.
- The *LS* factors presented in the RUSLEFAC are representative of straight slopes, but can be manipulated to represent complex slopes (i.e. convex, concave, slopes with benches).

C is the cover factor (dimensionless).

- *C* is *dependent* on the vegetative cover and the land use.
- This factor is based on tables available in the RUSLEFAC document (Wall et al., 2002).

P is the support practice factor (dimensionless).

- The support practice factor accounts for the effects of practices that may reduce the volume or rate of runoff water by altering the flow pattern, surface grade, or direction of surface runoff.

6 RUSLEFAC Inputs

To determine the impact and sensitivity of the input variables on soil loss, a range of values were used for each variable. The ranges of input values are discussed in the following subsections. The results of the analyses using the discussed ranges of input values are included in Section 7.

6.1 Erosivity/Rainfall Factor (R)

Annual erosivity represents the average precipitation energy that causes soil loss over the course of an average year. The annual erositivity value can be used to determine the cumulative soil loss over a long period of time.

Annual R values are not shown on the Canadian Isoerodent Maps in the Yukon Territory near the Site. As discussed in Section 5, erosivity is greatly dependent on rainfall. Therefore, to determine the erosivity at the site, SRK compared total annual precipitation as rainfall to erosivity in locations with known erosivities, then applied the trends to the site (annual precipitation as rainfall for the site was taken from Environment Canada's Pelly Crossing Station). Based on the application of the trends, the erosivity at the site likely falls within the range of 200 to 240 MJmm/hah. For conservatism, SRK applied a 25% contingency to the upper end of the range. The conservatively estimated R value for the site is therefore 300 MJmm/hah.

Soil loss estimates for short term periods (i.e. single storm events) were not included in this analysis.

6.2 Soil Erodibility Factor (K)

Figure 1 illustrates the particle size distribution data plotted in accordance with the USDA Soil Textural classification, and the general site area from which that sample was obtained. Figure 1 also illustrates the average of all samples (on which the majority of this analysis is completed), and the upper (maximum) and lower (minimum) bounds, discussed in further detail in Section 7.4. The material available for cover is generally classified as a sandy loam. Material properties from 70 soil samples were averaged and evaluated using the soil erodibility nomograph (Wall et al., 2002); the resulting K value was 0.027. Approximate minimum and maximum K values were then estimated using the soil erodibility nomograph, which were 0.011 and 0.051, respectively.

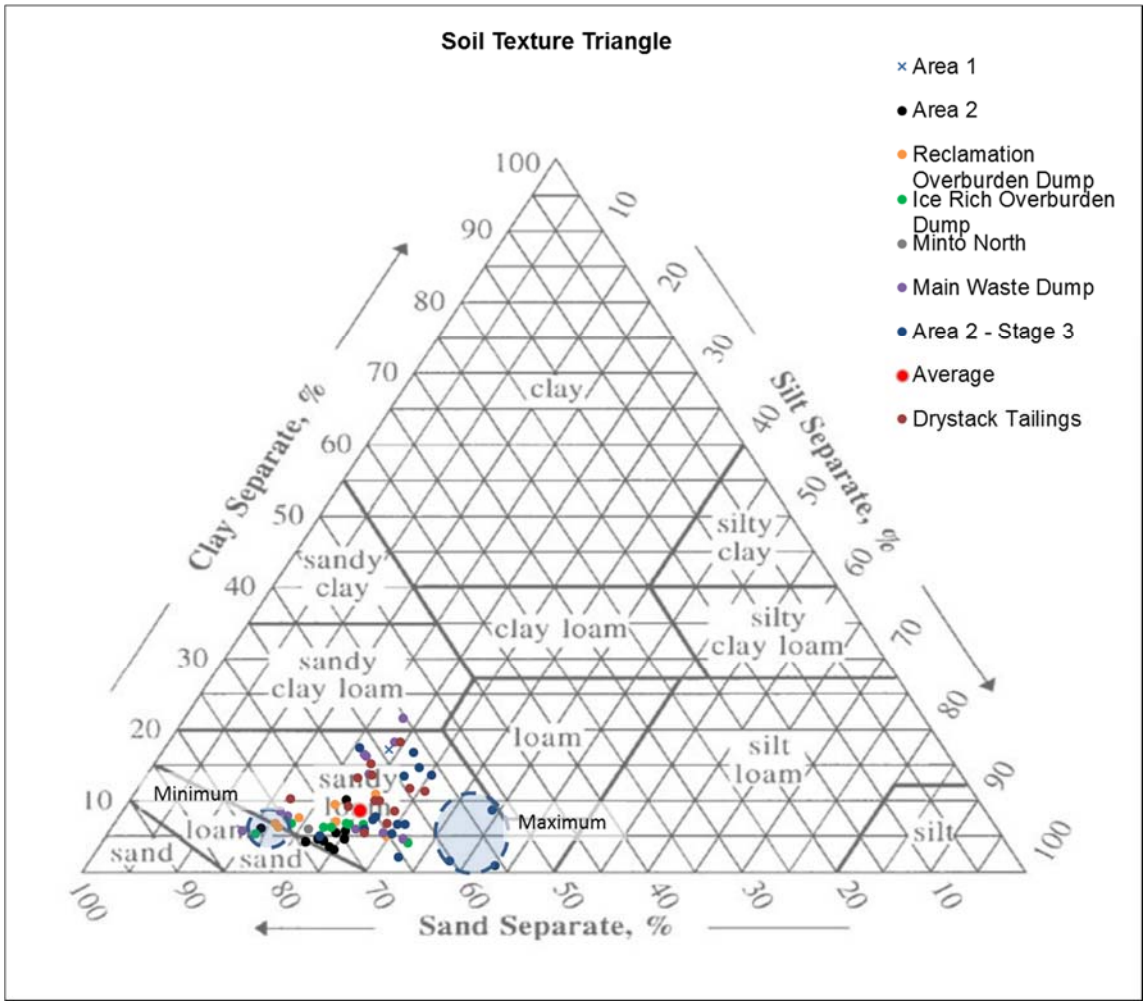


Figure 1: Soil Texture Triangle

6.3 Length and Slope Steepness Factors (L&S)

Several different straight and complex slopes were assessed. Straight slopes of 6H:1V, 5H:1V, 4H:1V, 3.5H:1V, 3H:1V, 2.5H:1V, and 2H:1V were each assessed for lengths of 10 m up to 200 m.

A variety of complex slopes were assessed that each had an average slope of 4H:1V and a length of 100 m. The complex slopes were assessed for the same length and slope to show the comparative difference between each type of slope. The complex slopes included four concave slopes (consisting of two to four straight segments), a straight slope with one 10 meter bench, and a straight slope with two 10 meter benches (the straight portions consisted of 4H:1V slopes, therefore the overall slope was substantially flatter than 4H:1V). The types of slopes that were assessed are illustrated in Figure 2. The figure indicates the horizontal to vertical slopes, but it is not drawn to scale.

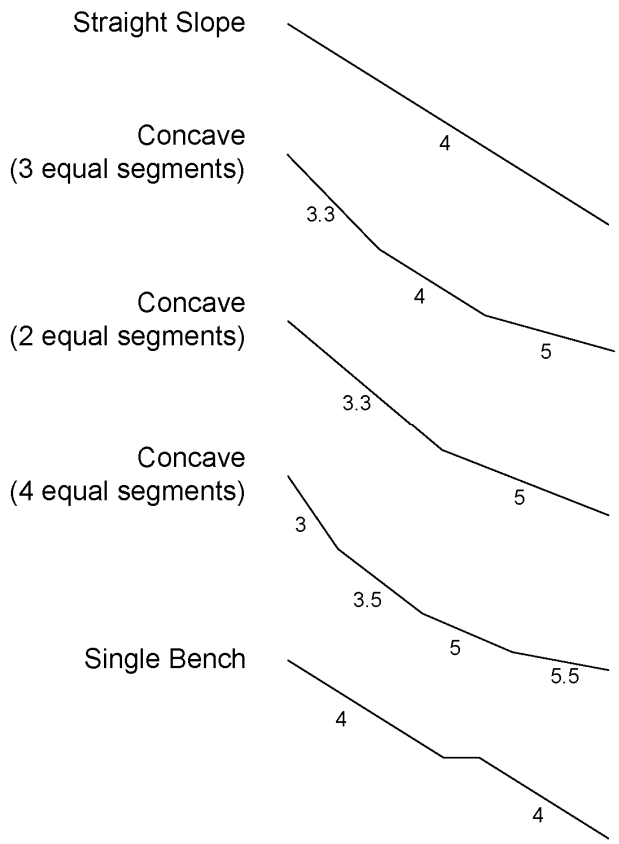


Figure 2: Types of Slopes Assessed

6.4 Cover Factor (C)

The *C* factor was determined using Table C-5 in the RUSLEFAC. Values decrease with lesser cover (yielding lesser soil loss). The value for bare, undisturbed soil with no vegetative canopy (canopy is considered having plants/weeds/shrubs of 0.5 m height or greater) or surface cover is 0.45. The value for 40% small, short-rooted plant coverage with no canopy is 0.15, and the value for 40% small, short-rooted plant coverage with a taller plant canopy is 0.13. Increasing small, short-rooted plant coverage to 80% with canopy decreases the cover factor to 0.04.

6.5 Support Practice Factor (P)

The base case P factor was to have no impact the on the soil loss equation and was made equal to one. The support practice factor is proportional to soil loss (i.e. a support practice factor of zero will yield zero soil loss).

Short term support practices could be incorporated into the design to support the process of establishing vegetation on the slopes. The support practices are likely to include slope texturing, sediment fencing (or other flow velocity reduction measures), and/or the use of rolled erosion control products. The respective support practice factors are 0.9, 0.6, and 0.1 respectively (Alberta, 2011). As stated, although not included as base case conditions, the effect of support practices was included as a sensitivity to demonstrate the impact in reducing soil loss.

7 Results and Discussion

The figures within this section show soil loss in units of tonnes per hectare per year (T/ha/year) and in millimeters per year (mm/year). The depth per year values were determined using an average dry density of 1.6 T/m³. The depth represents the average depth of soil loss over the entire erodible surface area. The guideline values for the Class 1 soil erosion class of 6 T/ha/year corresponds to a depth of 0.35 mm/year. The guideline values are not shown on Figures 5, 6, and 7, as these figures are intended to show the relative difference of how certain parameters affect erosion, and were not necessarily intended to show the design slopes that will be selected at the site.

7.1 Straight Slopes

Figure 3 illustrates the expected straight slope soil loss with the average available material if no vegetative cover is established. None of the scenarios meet the target of 6 T/ha/year.

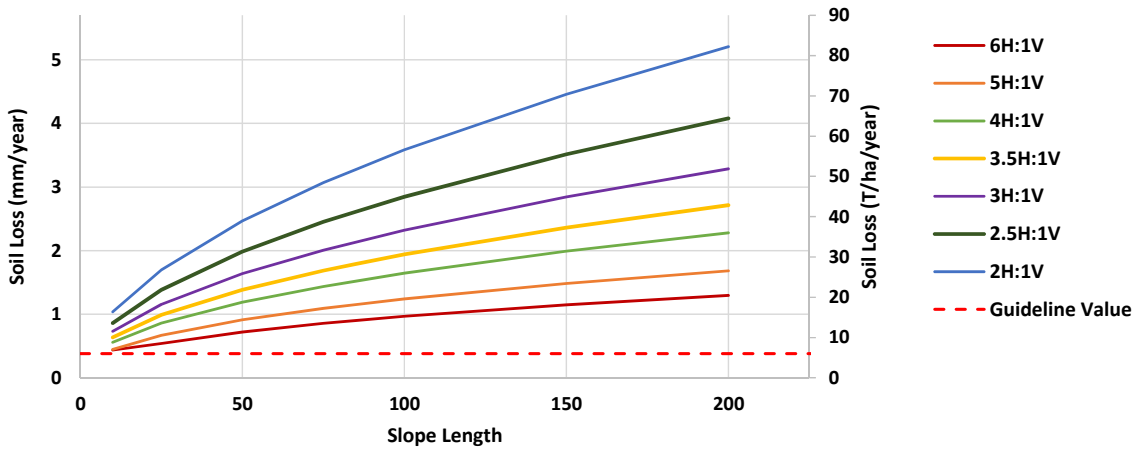


Figure 3: Straight Slopes using Average Material with no Vegetative Cover

7.2 Effects of Vegetation

Figure 4 illustrates the expected straight slope soil loss with 80% small, short-rooted plant coverage and no vegetative canopy. Comparing Figure 3 and Figure 4 shows that established vegetation significantly reduces soil loss due to water erosion. Most of the assessed slope conditions meet the target of 6 T/ha/year.

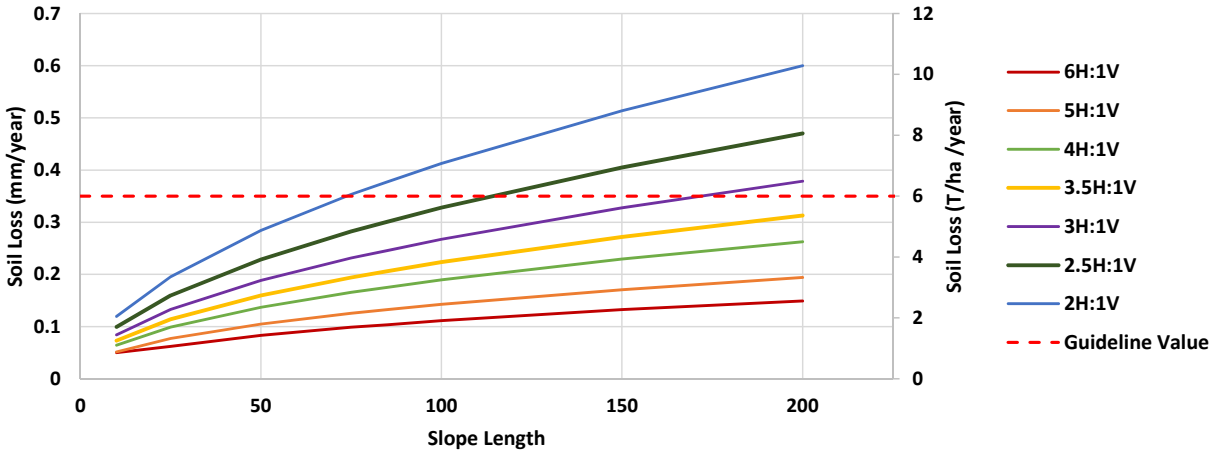


Figure 4: Straight Slopes using Average Material with 80% Small, Short-rooted Plant Coverage and No Vegetative Canopy

Figure 5 shows the effects that increased vegetation coverage have on a particular slope. The figure shows that achieving at least some vegetation coverage (20%) reduces soil loss due to erosion by a significant margin.

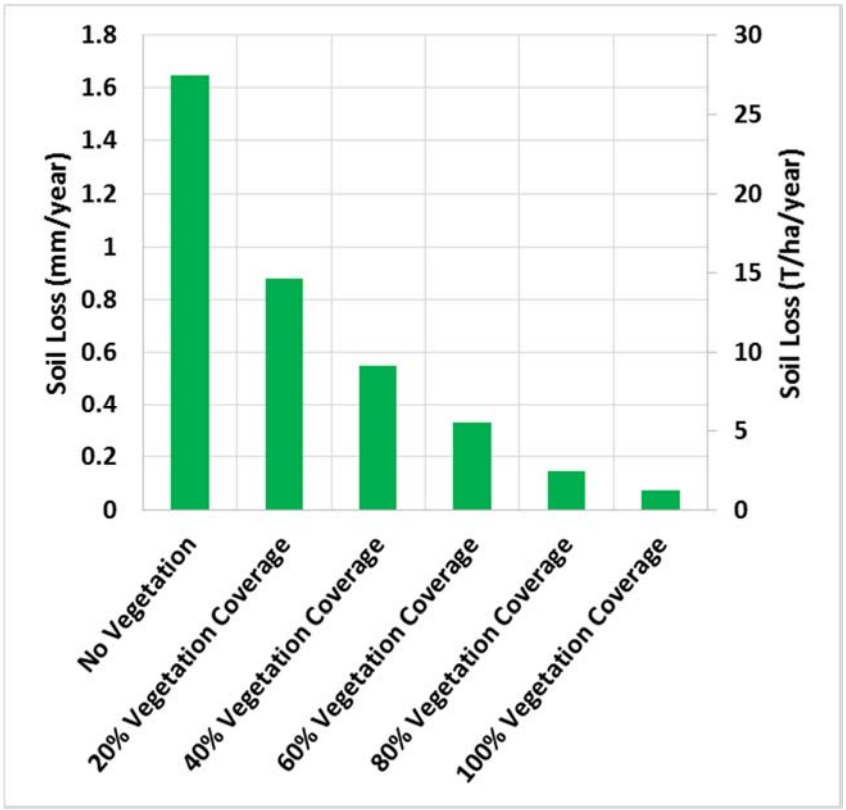


Figure 5: The Effects of Vegetation Coverage on a 4H:1V, 100 m Slope with Average Material

7.3 Effects of Complex Slopes

The soil losses for 100 m long complex slopes at 4H:1V with no vegetative cover and average site material are shown in Figure 6. The figure indicates that each of the complex slopes yields less soil loss than an equivalent straight slope. Complex slopes were somewhat effective at reducing soil loss in this analysis: soil loss was approximately 9% less on concave slopes than on straight slopes. Although only 100 m, 4H:1V slopes are presented, SRK has determined via the RUSLEFAC, the reduction in soil loss on complex slopes is similar for other slopes and slope lengths in the same order of magnitude (i.e. 5H:1V slopes, 50 to 125 m slope lengths). The soil loss reductions are expected to be less similar to those presented if the slope length or steepness is increased substantially.

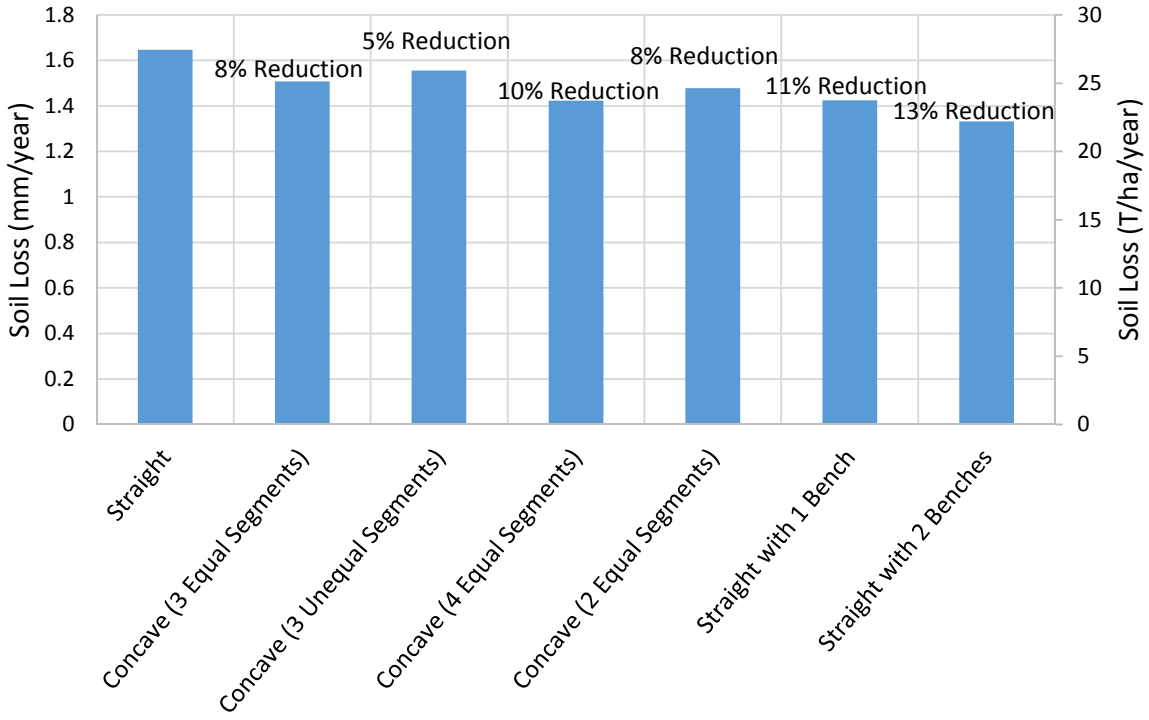


Figure 6: Complex Slope Comparison (100 m Long at 4H:1V and No Vegetative Cover)

7.4 Effects of Soil Type

The effects of soil type are presented in Figure 7. Each of the soil loss estimates are based on 100 m long 4H:1V straight slopes, and no vegetative cover. The figure indicates the range in erosion susceptible material available on site. It is important that material susceptibility to erosion be considered in design stage of the cover. By choosing material that is more susceptible than the average material available on site, erosion estimates can increase by as much as 100%. More erosion susceptible material contains a greater percentage of silt and fine sand, while less erodible material contains less silt and fine sand. The classification of these soils is described in Section 6.2.

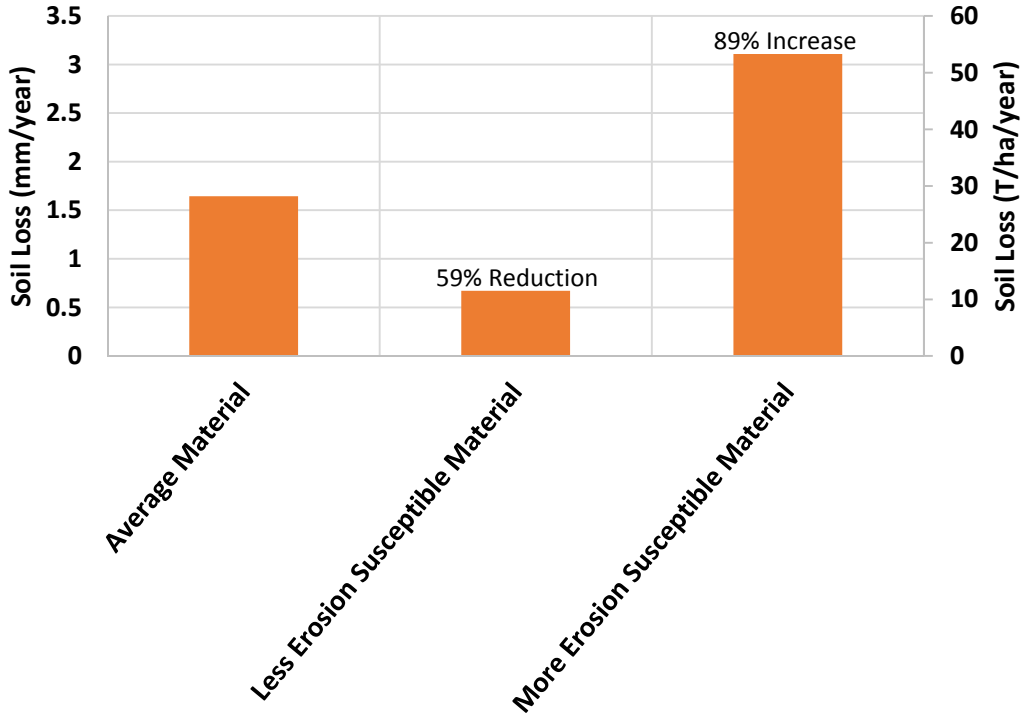


Figure 7: Soil Type Comparison (Based on 100 m Long 4H:1V Straight Slopes)

7.5 Effects of Support Practice Factor

The effects of the support practice factor were evaluated on non-vegetated, 100 m long, 4H:1V slope, covered with average material available on site. The results are presented in Figure 8. The figure shows that through the use of support practices, also commonly referred to as the incorporation of microtopography, the estimates of erosion can be decreased to the target of 6 T/ha/year even without the establishment of vegetation. The use of soil texturing alone will not reduce the rate of erosion to the target; rolled erosion control products would be required to meet the target without vegetation. Sediment fencing is grouped together with wattles as velocity reducers, as their effectiveness in reducing erosion is similar.

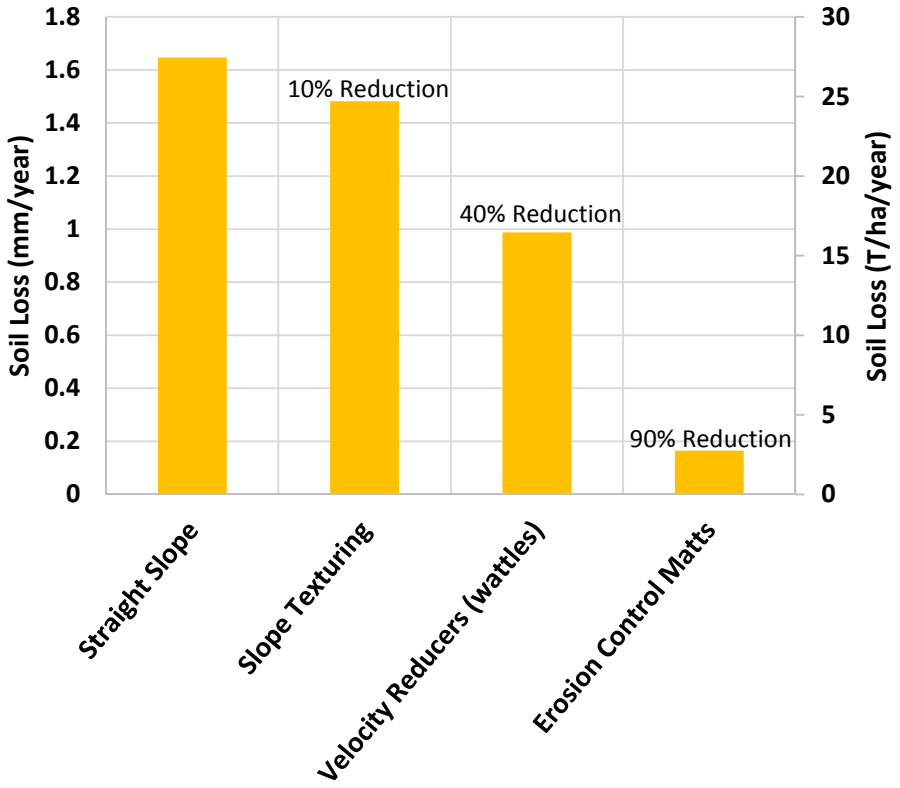


Figure 8: Support Practice Comparison (No Vegetation)

8 Total Soil Loss

Soil loss over the course of the design life was calculated to determine whether the average depth of soil loss would reduce the initial cover thickness to below the required cover thickness. Annual soil loss due to water erosion was multiplied by 100 years to determine design life soil loss, which is presented for several straight slope scenarios in Table 8-1. Average annual soil loss (in T/ha/year) is also presented in the table.

Table 8-1: Calculated Water Erosion Design Life Soil Loss

Slope Condition		Design Life Soil Loss (mm) per Slope Length							
		50 m		85 m		100 m		150 m	
		Annual (T/ha/yr)	100 yrs (cm)	Annual (T/ha/yr)	100 yrs (cm)	Annual (T/ha/yr)	100 yrs (cm)	Annual (T/ha/yr)	100 yrs (cm)
Non-Vegetated	**2.5H:1V**	31.8	19.9	41.8	26.1	45.6	28.5	56.2	35.1
	3H:1V	26.2	16.4	34.1	21.3	37.1	23.2	45.5	28.4
	3.5H:1V	22.1	13.8	28.6	17.9	31.0	19.4	37.8	23.6
	4H:1V	19.0	11.9	24.3	15.2	26.4	16.5	31.9	19.9
	5H:1V	14.6	9.1	18.4	11.5	19.8	12.4	23.7	14.8
Vegetated (80% Short-Rooted Plant Coverage)	**2.5H:1V**	3.7	2.3	4.8	3.0	2.3	3.3	6.5	4.1
	3H:1V	3.0	1.9	3.9	2.5	4.3	2.7	5.2	3.3
	3.5H:1V	2.6	1.6	3.3	2.1	3.6	2.2	4.4	2.7
	4H:1V	2.2	1.4	2.8	1.8	3.0	1.89	3.7	2.3
	5H:1V	1.7	1.1	2.1	1.3	2.3	1.4	2.7	1.7

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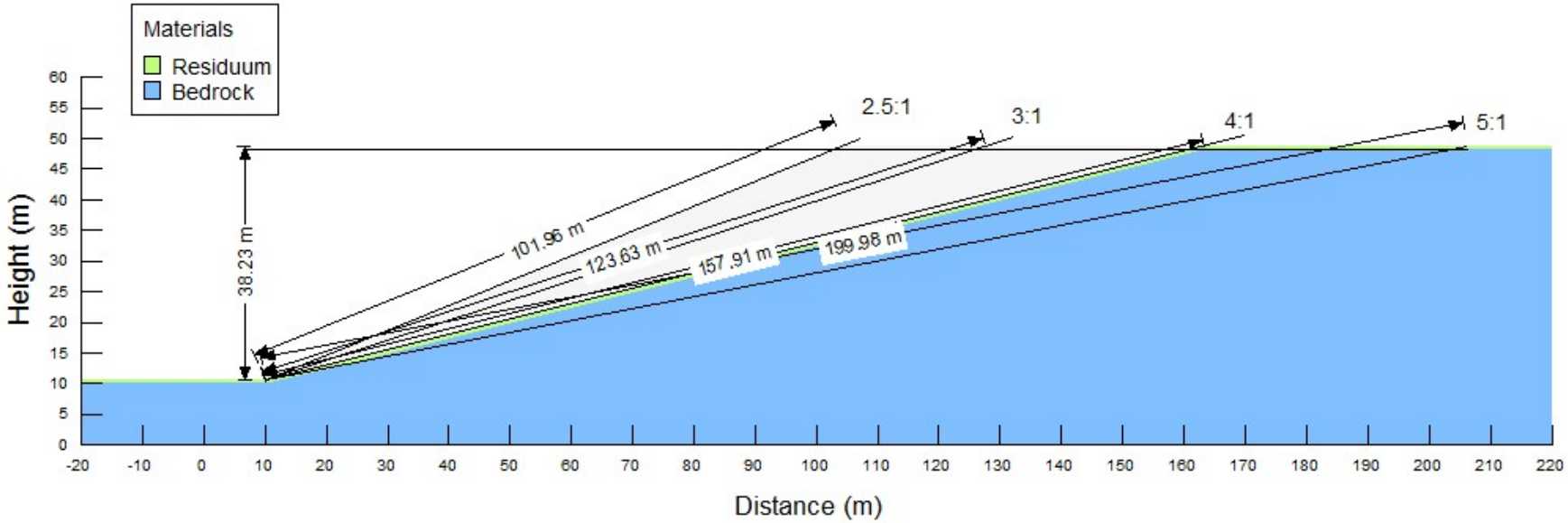
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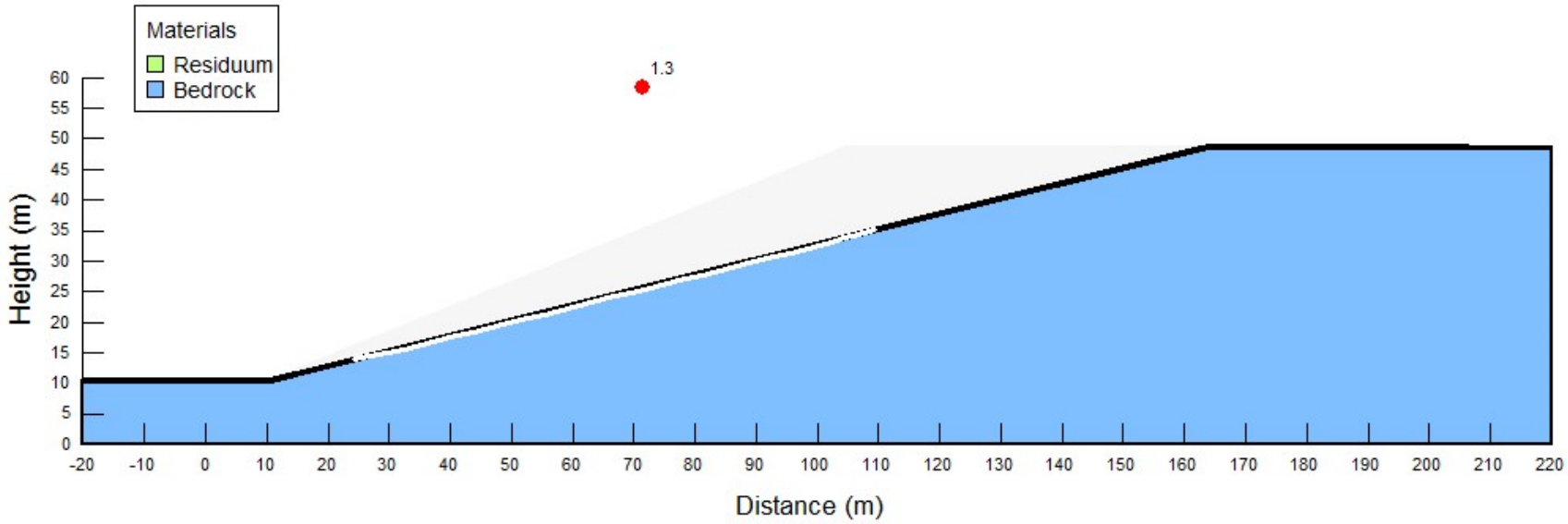
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Appendix D – Minto Closure Cover Design – Stability Assessment

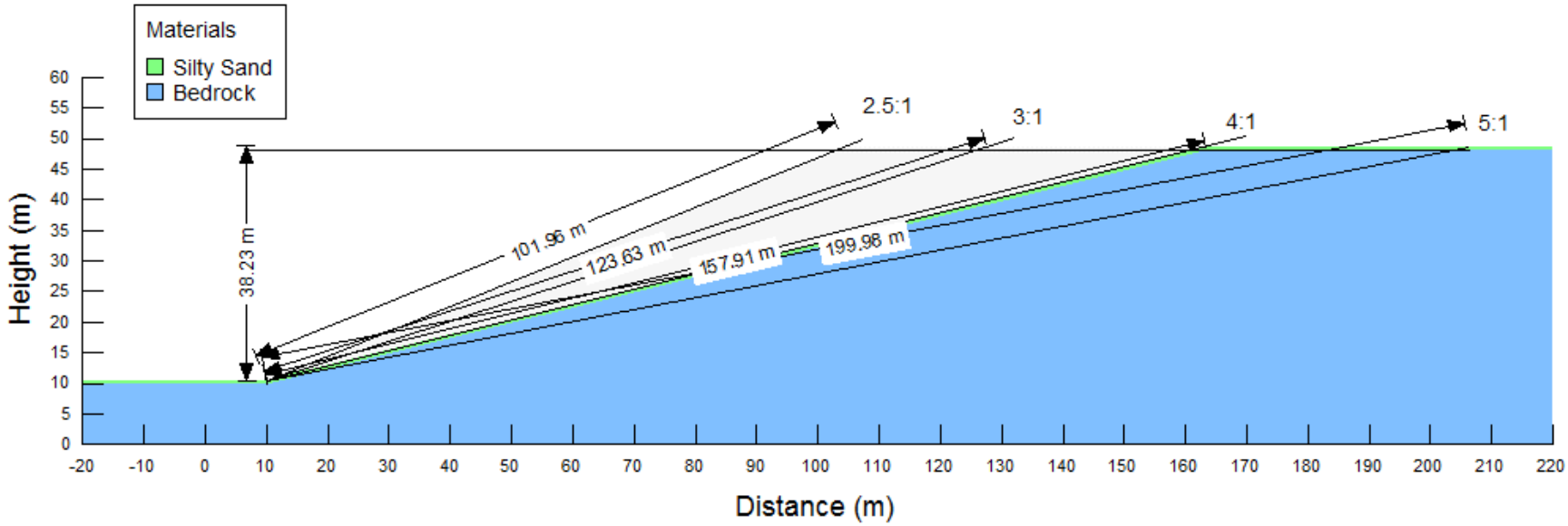


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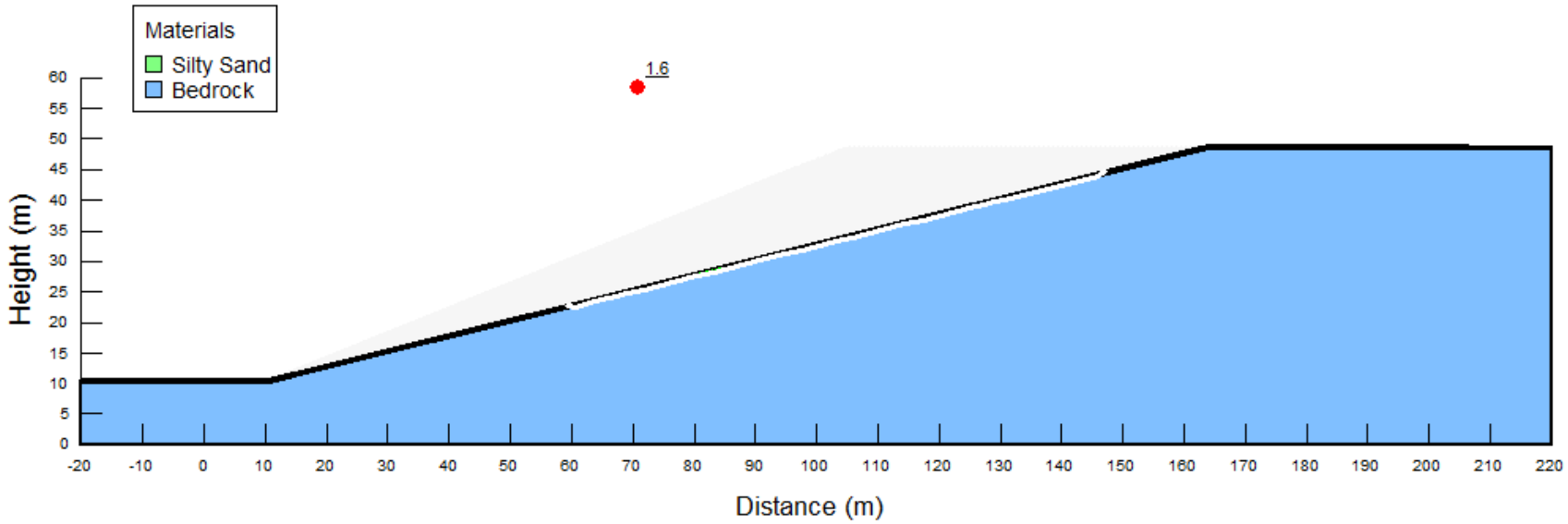


Results

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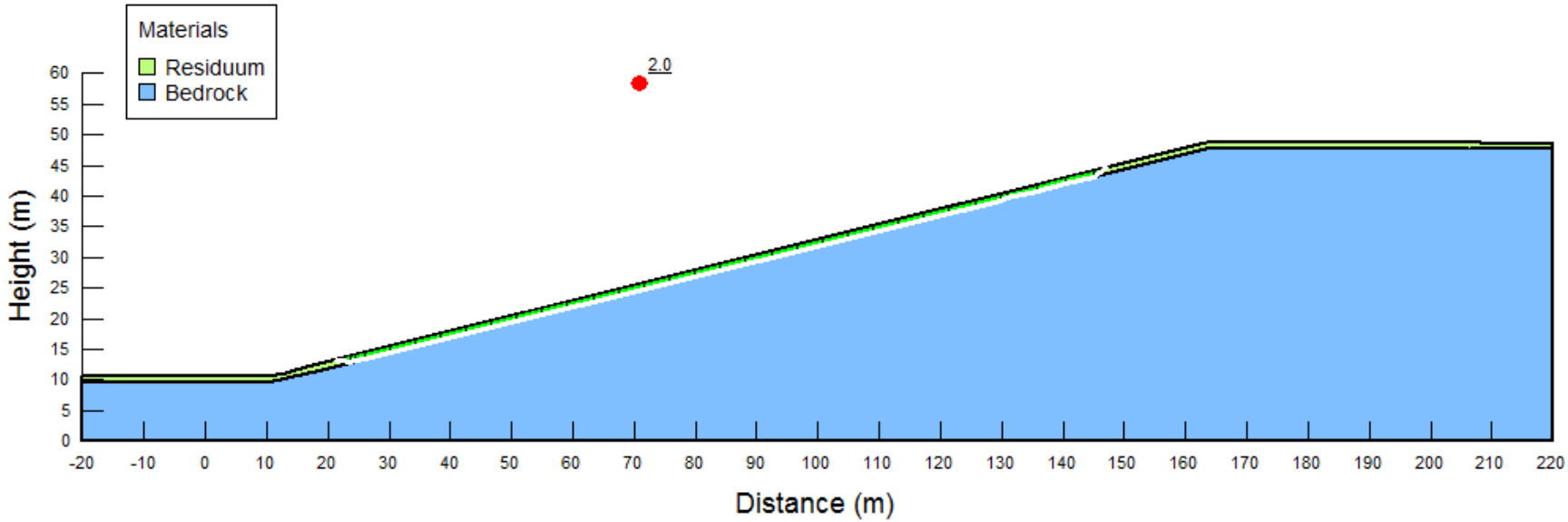


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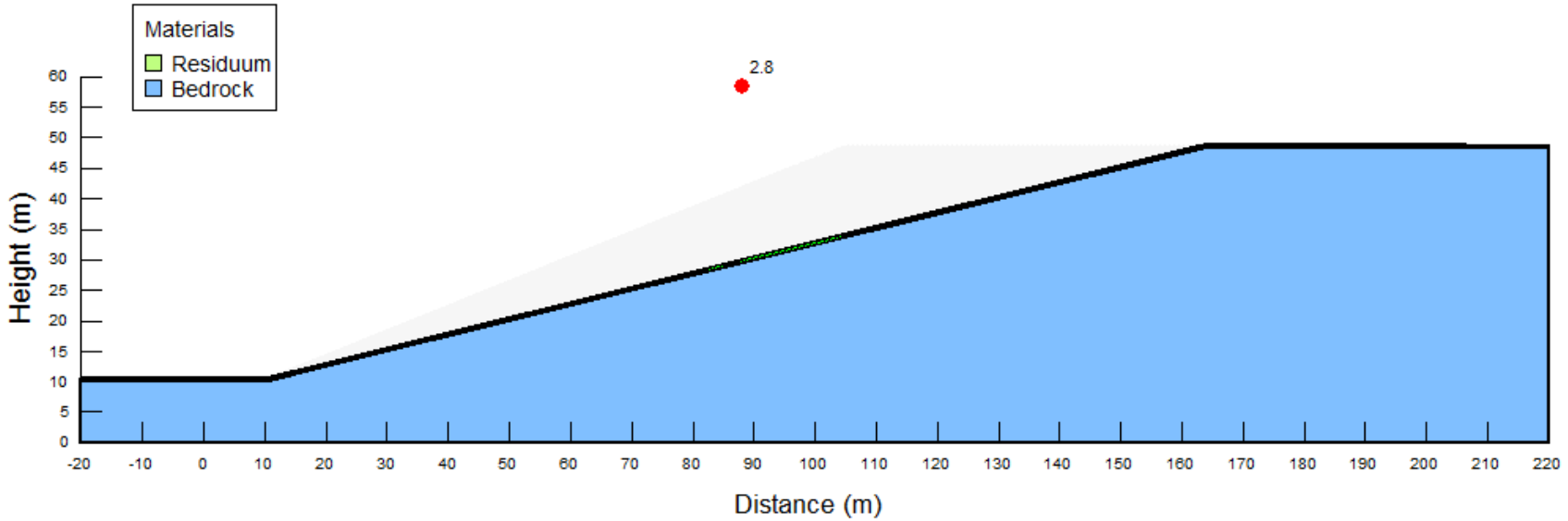


Results

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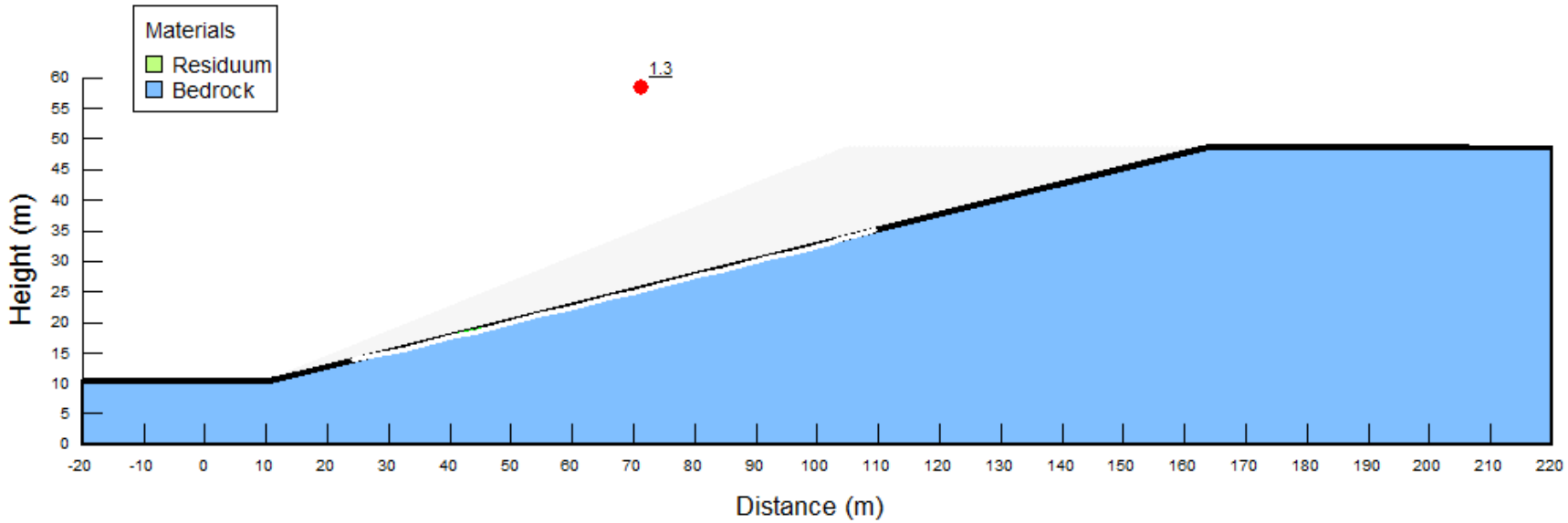


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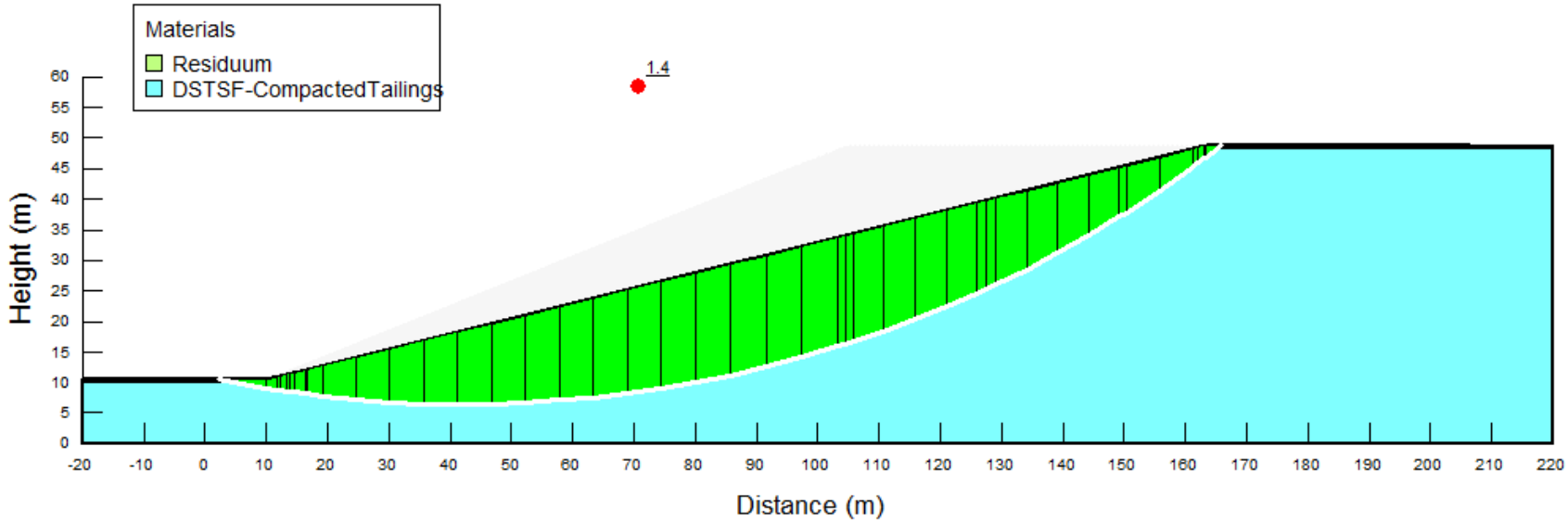


Scenario 1b

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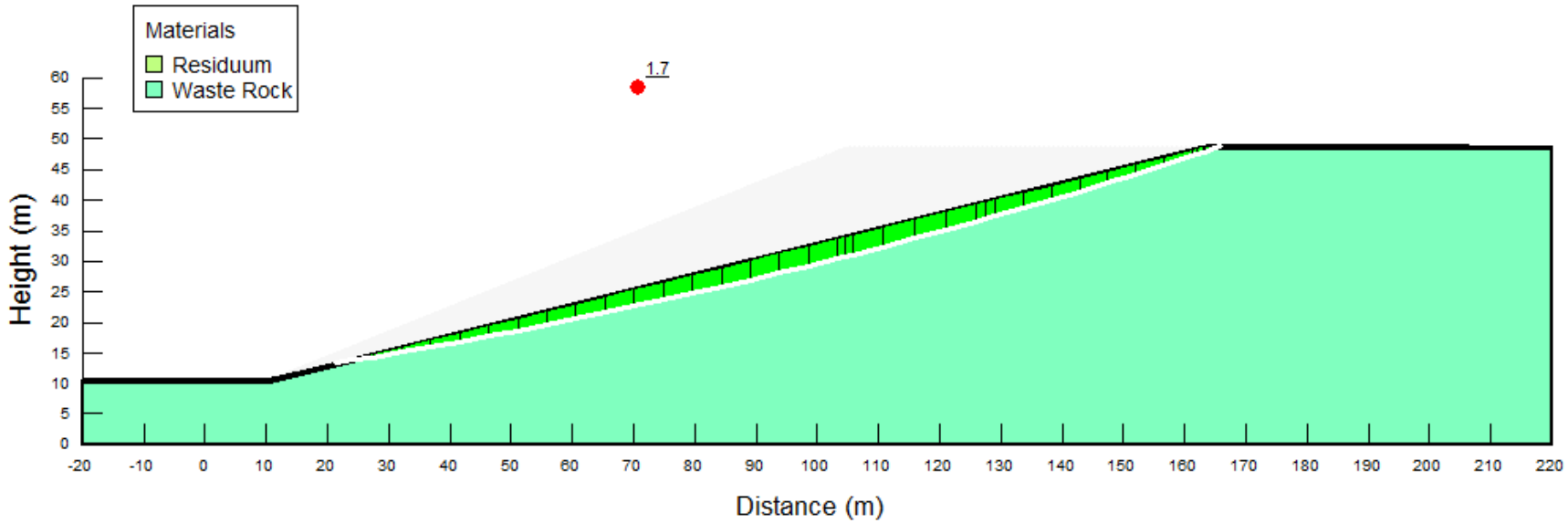


Scenario 1c

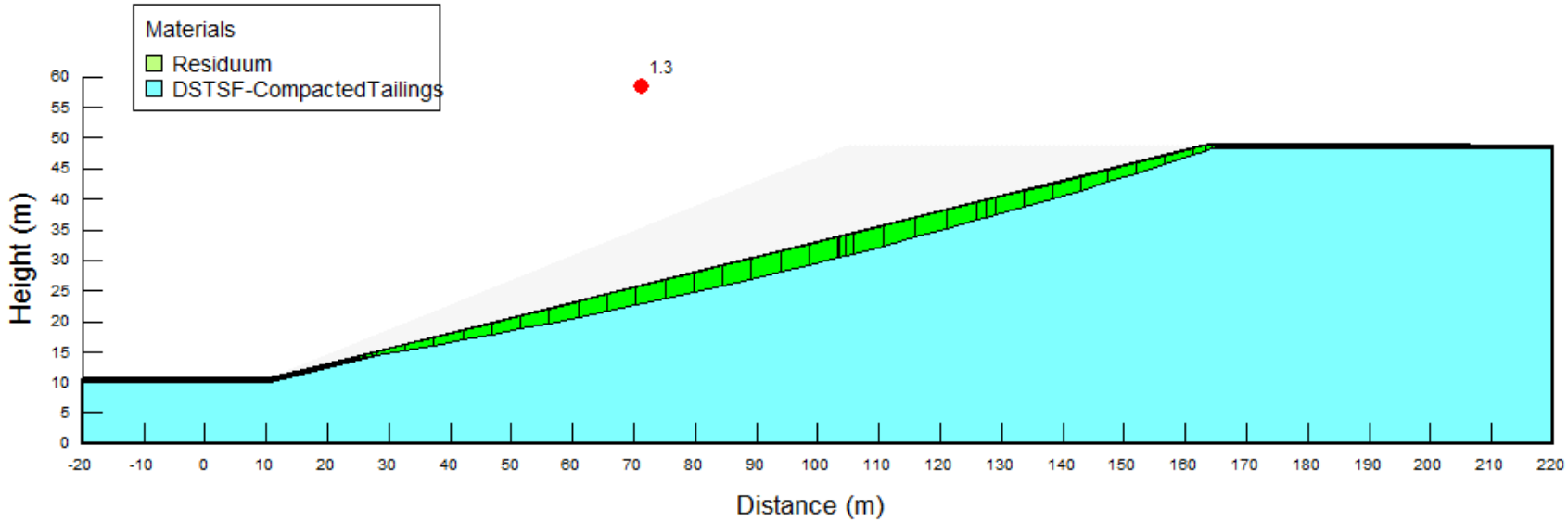


Scenario 1d

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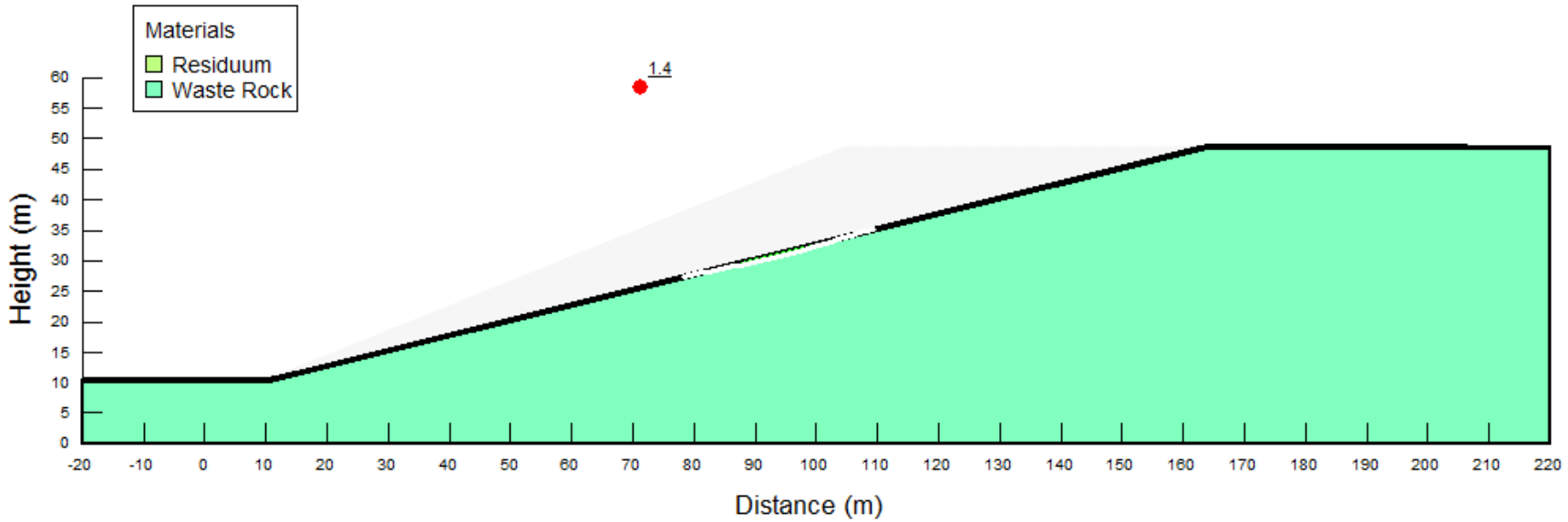


Scenario 1e

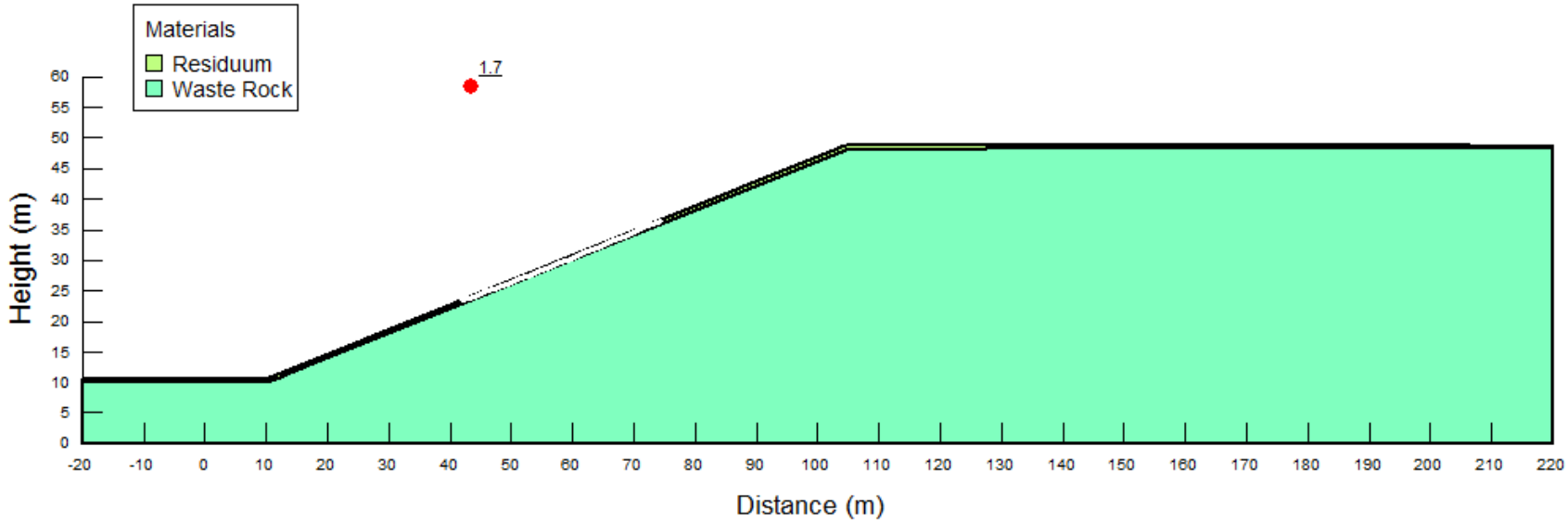


Scenario 1f

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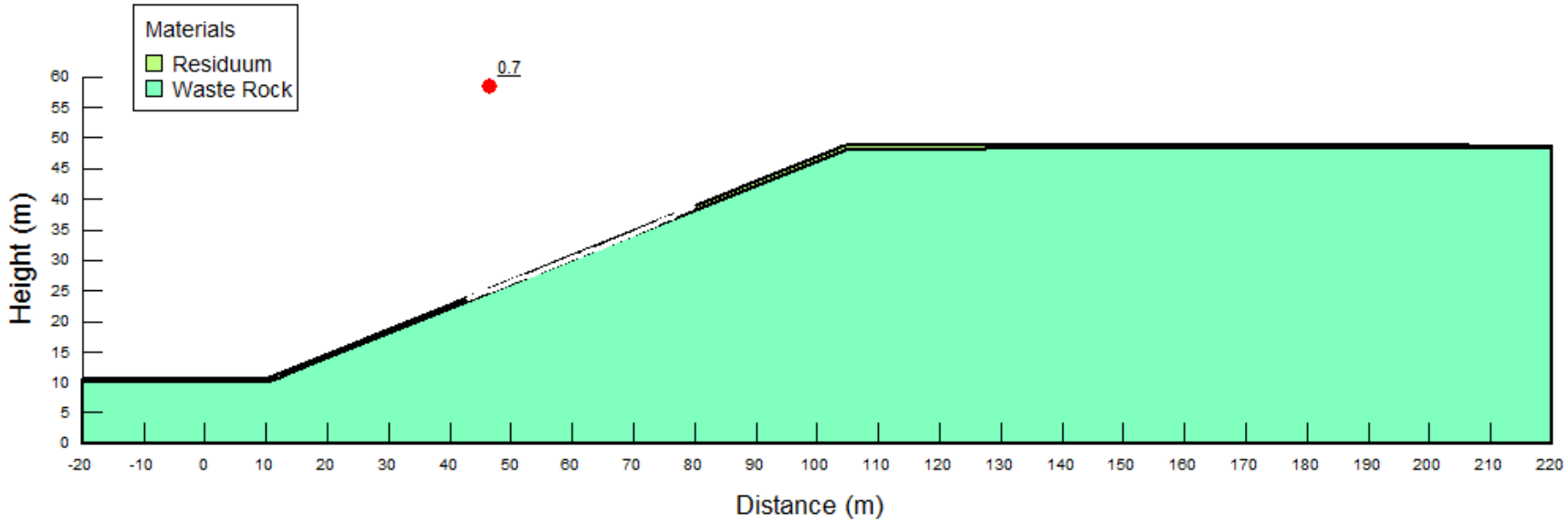


Scenario 1e



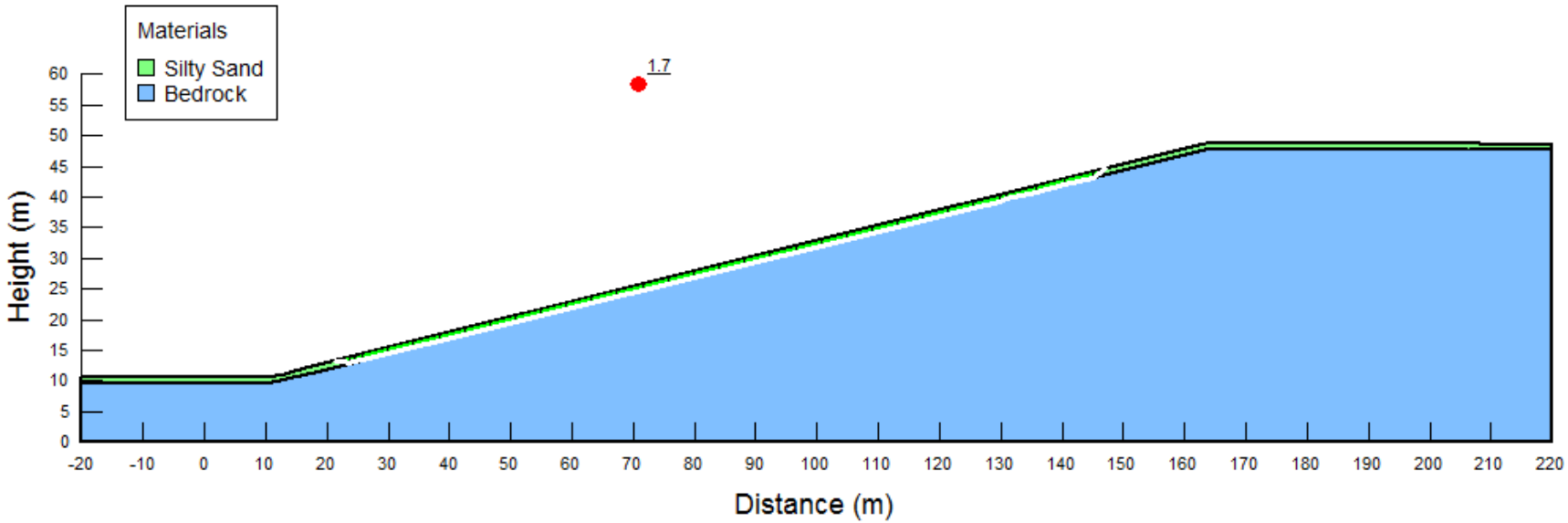
Scenario 1f

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		Date: July 2016 / Approved: EK / Figure: **6**

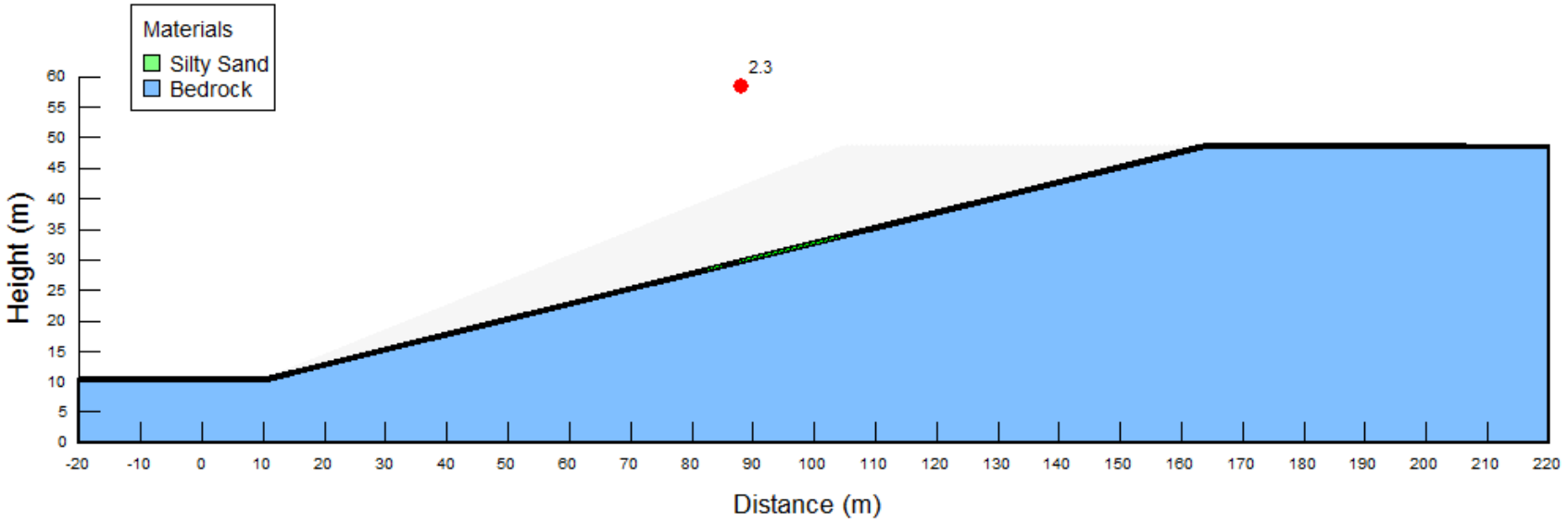


Scenario 1i

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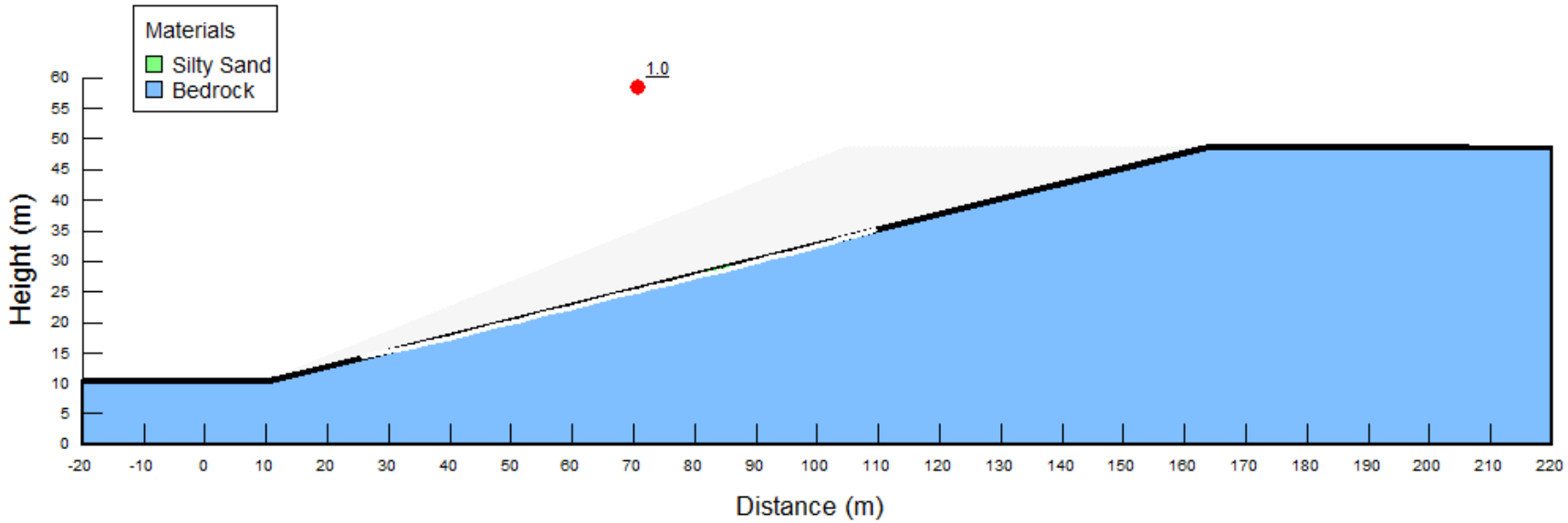


Scenario 2a

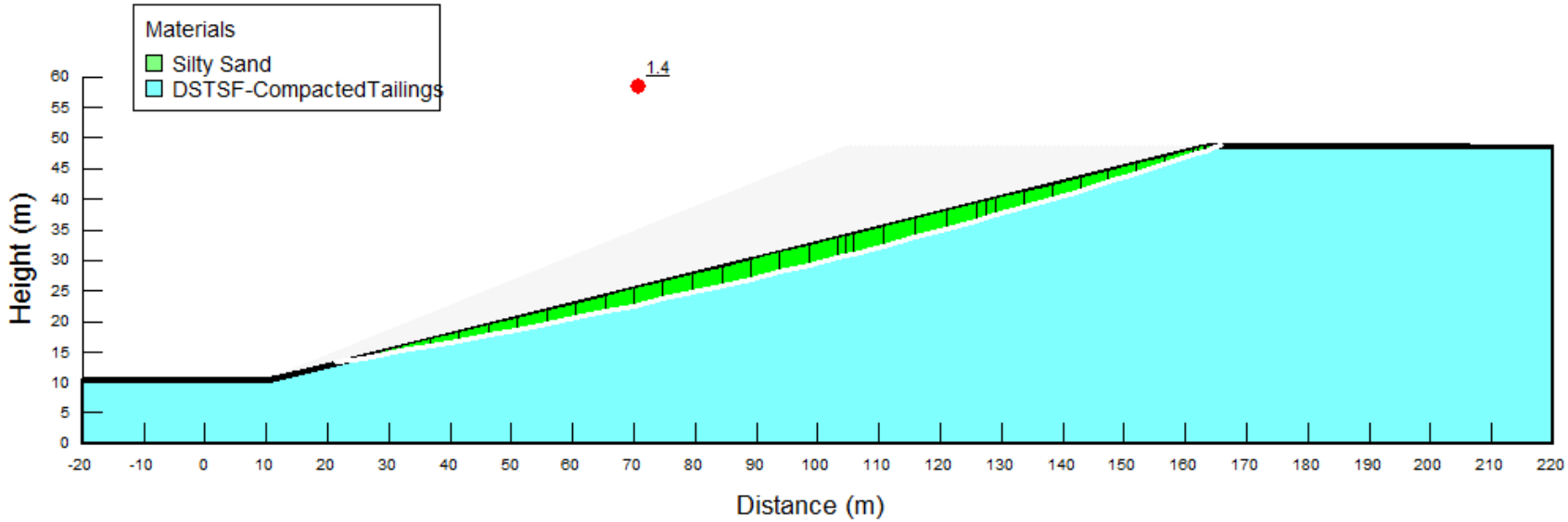


Scenario 2b

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		Scenario 2 Sensitivity
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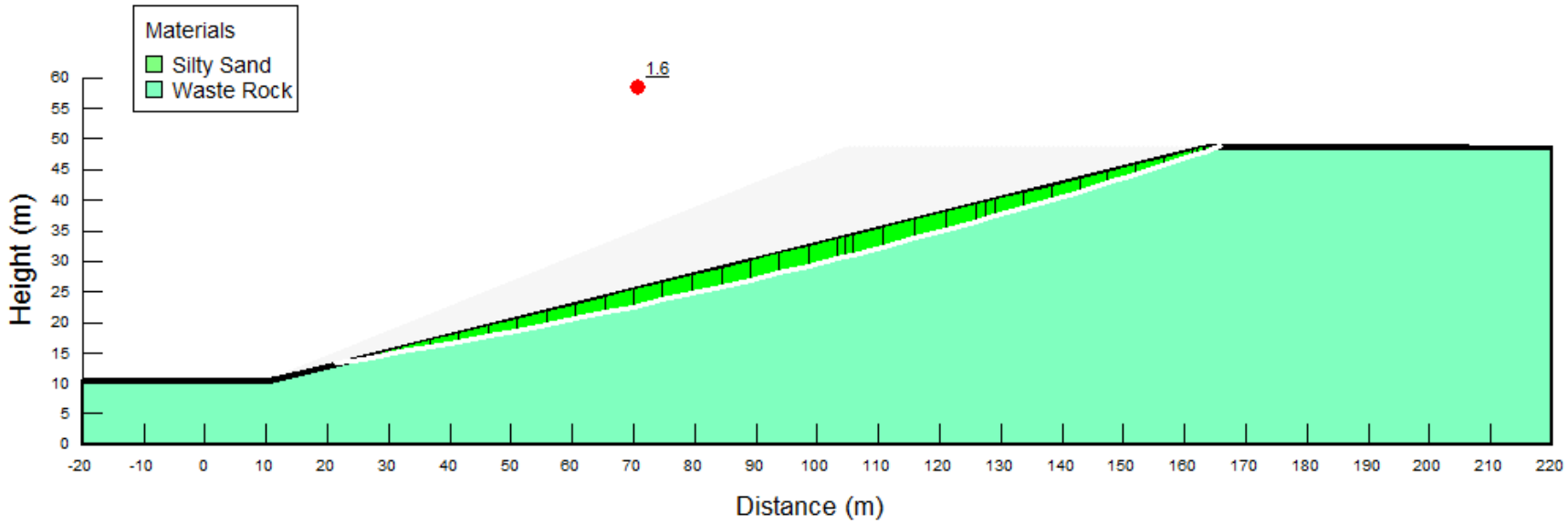


Scenario 2c

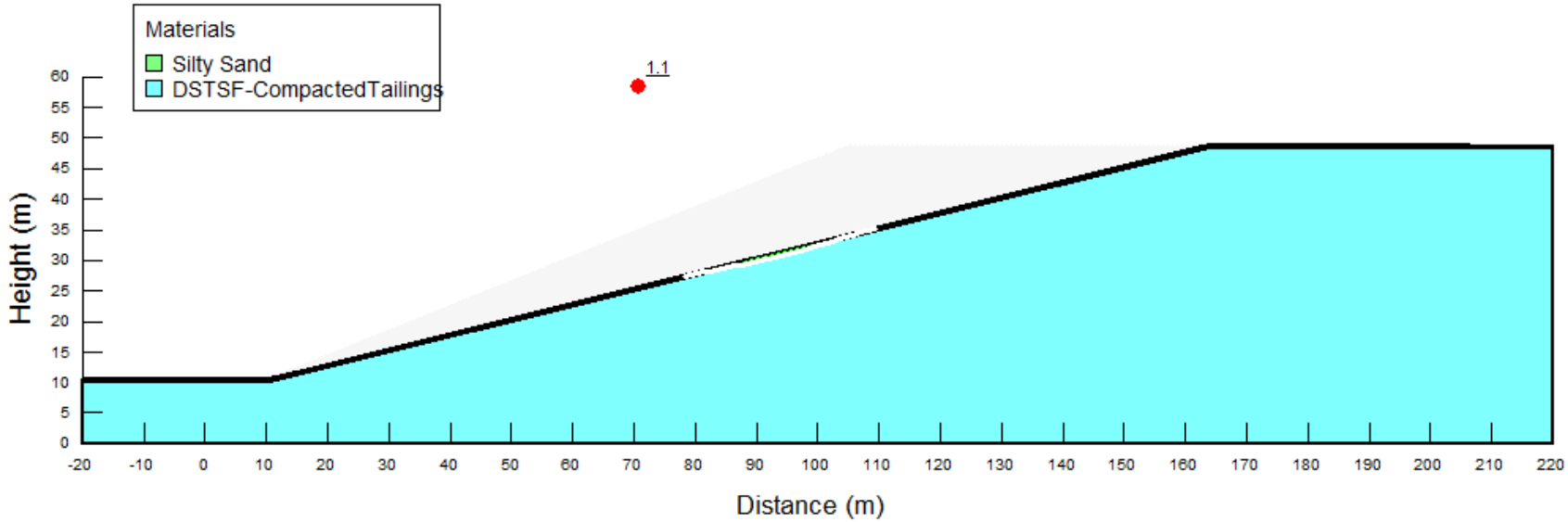


Scenario 2d

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		Date: July 2016 / Approved: EK / Figure: **9**

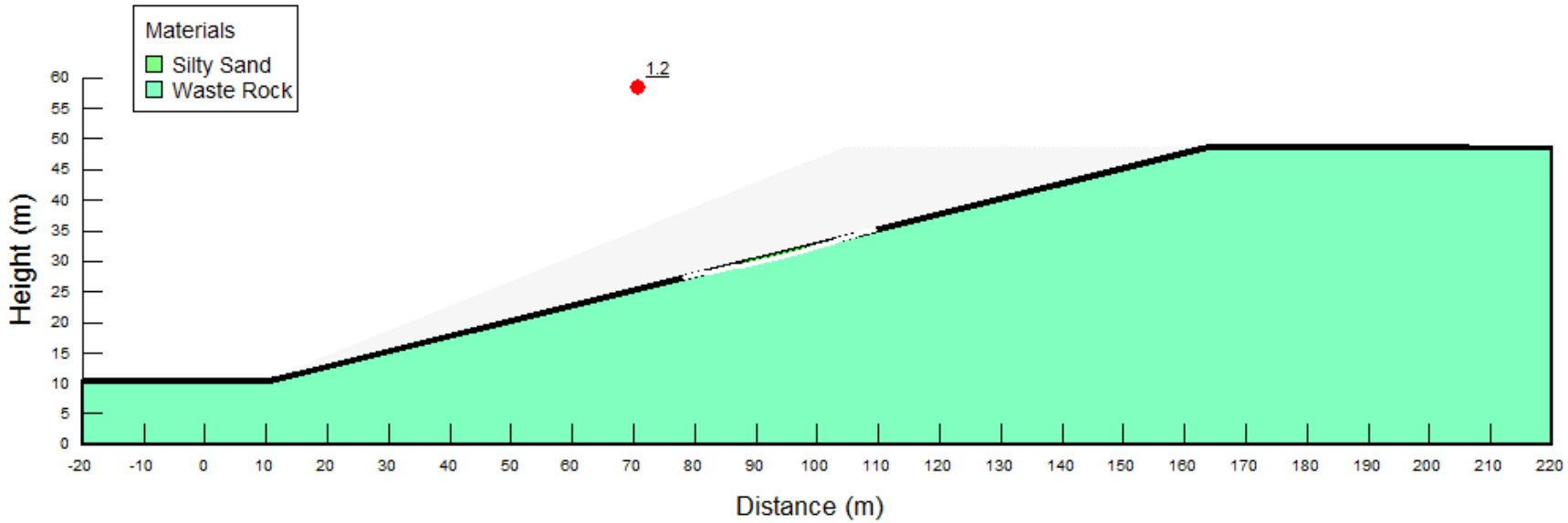


Scenario 2e

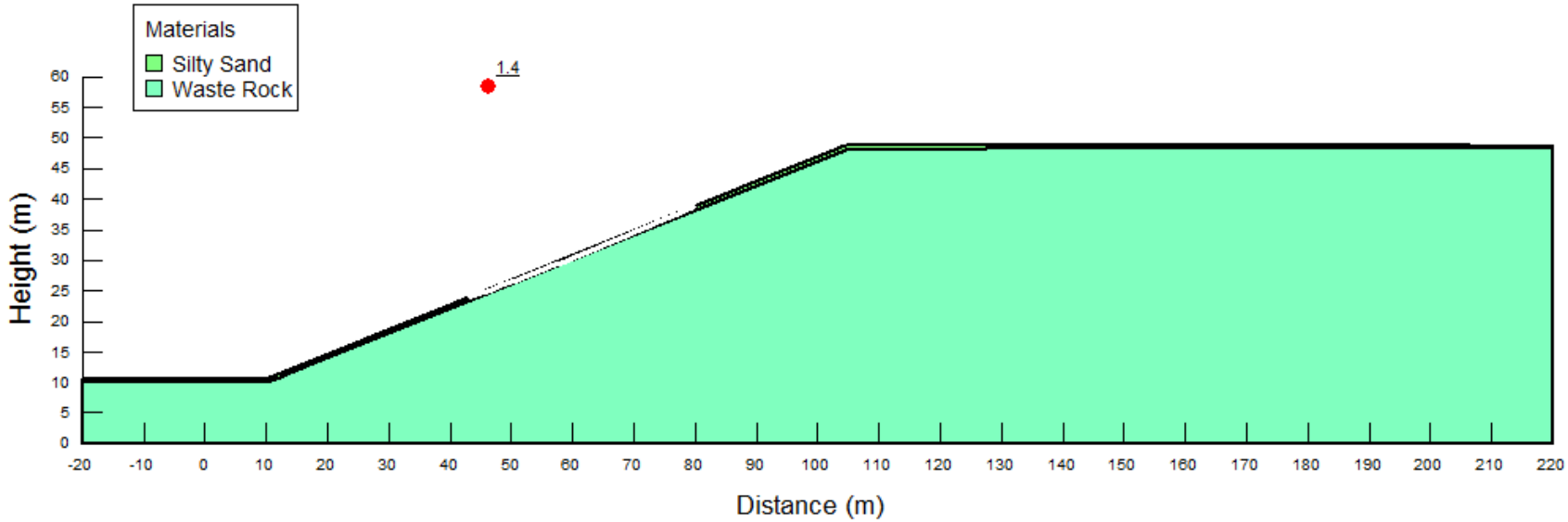


Scenario 2f

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		Date: July 2016 / Approved: EK / Figure: **10**

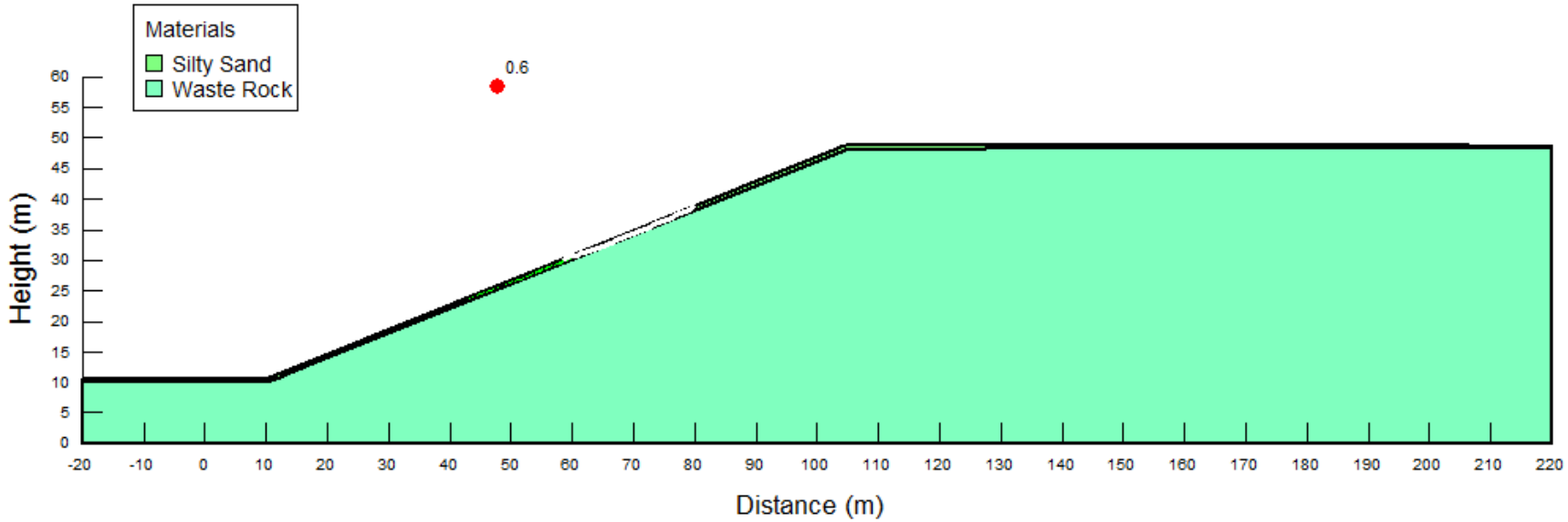


Scenario 2g



Scenario 2h

srk consulting	capstone MINTO MINE OPERATED BY MINTO EXPLORATIONS LTD.	Stability Assessment
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		Date: July 2016 / Approved: EK / Figure: **11**



Scenario 1i

srk consulting	Capstone MINTO MINE OPERATED BY MINTO EXPLORATIONS LTD.	Stability Assessment	
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		Date: July 2016 / Approved: EK	Figure: **12**

Appendix E – Minto Mine Preliminary Closure Cover Revegetation Plan

IEG
Integral Ecology Group
iegconsulting.com

MEMORANDUM

December 3, 2017

TO:	Erik Ketilson, SRK Consulting (Canada) Inc.
FROM:	Justin Straker
RE:	Update on revegetation planning for the Minto mine site, 2017
COPY:	

Please find below an update on activities completed with respect to revegetation planning at Minto in 2017. This update is not comprehensive, as some of our activities are still in progress, and will be reported later.

Site visit

A site visit was conducted on July 6, 2017, with Ryan Herbert (Capstone/Minto), Erik Ketilson and Dylan MacGregor (SRK), and Justin Straker (IEG) in attendance. The site visit focussed on surveying:

- non-mine areas as references for revegetation targets,
- established revegetation trials on Main Waste Dump (MWD);
- areas of natural regeneration on overburden stockpiles; and
- areas on the Dry-Stack Tailings Storage Facility (DSTSF) that are ready or nearly ready for revegetation.

Development of seed mixes for Minto for 2017/2018

SRK and Minto provided mapping of reclamation-ready or near-reclamation-ready areas for revegetation in 2017 and 2018. This mapping was then stratified as follows:

1. slopes versus plateaus – a primary goal of revegetation is to reduce erosion, where needed. Thus we identified sloped areas for erosion-control revegetation treatments and level areas where these treatments are not required; and
2. zonal(mesic) sites versus drier (subxeric-submesic) sites – the former occur on tailings substrates and north-facing slopes, while the latter occur on waste-rock substrates and south-facing slopes. Additional description of this mapping process is provided below.

An initial (rough) map of these revegetation classes is provided as Appendix A. Nomenclature for this map uses an adaptation of ecosites as described by Environment Yukon (2016 and 2017), where zonal sites are labelled as a site type containing the 01 and 30 ecosites (30 / 01), and drier sites are labelled as the 10 ecosite.

1% FOR THE PLANET

This mapping indicated that approximately 42 ha of level/plateau areas and 38 ha of sloped areas could potentially be available for revegetation in the fall of 2017 or 2018. Seed mixes were designed for these areas based on the following principles, with the overarching objective that revegetation materials should be wholly or mostly made up of species native to the Yukon:

- **sloped areas** – the primary revegetation goal on these areas is erosion control, so the seed mix should be primarily composed of native grass species, seeded relatively heavily (35 kg/ha), with an additional component of annual ryegrass and alfalfa, to accelerate revegetation and reduce the amount of bare ground exposed to erosion. Alfalfa was included as it is the only native legume still listed as an acceptable species by the Yukon Revegetation Manual (Matheus and Omtzigt 2013).
- **level areas** – the primary revegetation goal on these areas is promotion of recovery towards pre-development ecosystems. For this reason, the primary revegetation strategy is natural regeneration, as this has been observed on many sites at Minto, and as the relatively small development footprint and proximity of non-mined seed sources facilitates this recovery mechanism. However, observations on the MWD revegetation trial and the naturally revegetating overburden stockpiles suggest that foxtail barley is present and readily able to colonize sites. This species is undesirable as a large component of revegetation, as it can cause problems for grazing wildlife after seed formation, as the long seed awns can get stuck in the mouth, nose, and eyes of grazing animals, potentially causing irritation and infection. Therefore a seed mix was developed for level areas at Minto, with the goal of providing some occupation of space to prevent full colonization by foxtail barley, but also of leaving space for natural colonization of these sites. The level-ground seed mix is composed solely of grass species native to the Yukon, applied at a rate lighter (20 kg/ha) than that used for the sloped areas.

Seed mixes were developed based on observations of species successful at the MWD, on similar observations from revegetation of the Grum Sulphide Cell at the Faro mine, and on recommendations from the seed supplier. Details on these seed mixes are found in the tables below.

Slope Seed Mix	% by Weight	Seeds/ lb	% by Seed Count
Agrostis scabra, **Ticklegrass**	0.50	4,000,000	7.45
Elymus glaucus, **Blue Wildrye (Smooth)**	15.00	131,000	7.32
Elymus lanceolatus, **Northern Wheatgrass**	15.00	167,000	9.33
Elymus trachycaulus, **Slender Wheatgrass**	25.00	145,000	13.50
Festuca saximontana, **Rocky Mountain Fescue**	10.00	679,000	25.29
Lolium multiflorum, **Annual Ryegrass (Diploid)**	20.00	217,000	16.17
Medicago sativa, **Alfalfa, Creeping Rooted**	12	226,798	10.14
Poa alpina, **Alpine Bluegrass**	2.00	1,000,000	7.45
Trisetum spicatum, **Spike Trisetum**	0.50	1,800,000	3.35
Total	100.000		100.00

The slope seed mix sown at 35 kg per hectare will spread approximately 200 seeds/ft^2.

Level Seed Mix	% by Weight	Seeds/ lb	% by Seed Count
Agrostis scabra, **Ticklegrass**	1.00	4,000,000	13.64
Elymus glaucus, **Blue Wildrye (Smooth)**	25.00	131,000	11.17
Elymus lanceolatus, **Northern Wheatgrass**	22.00	167,000	12.53
Elymus trachycaulus, **Slender Wheatgrass**	35.00	145,000	17.30
Festuca saximontana, **Rocky Mountain Fescue**	14.00	679,000	32.41
Poa alpina, **Alpine Bluegrass**	2.00	1,000,000	6.82
Trisetum spicatum, **Spike Trisetum**	1.00	1,800,000	6.14
Total	100.000		100.00

The level seed mix sown at 20 kg per hectare will spread approximately 125 seeds/ft^2.

Due to the multi-species nature of the seed mixes, and the amplitude of tolerance to soil-water conditions of the species in the mixes, different seed mixes are not proposed for zonal versus drier areas. Different treatments for these areas will be used any planting programs, as described below.

The seed mixes are designed to be co-applied with an NPKS fertilizer (18-18-18-1), with 50% of the nitrogen provided in a slow-release form (released over a 30 to 60 day period). Fertilizer was designed for an application rate of 125 kg/ha on sloped areas and 70 kg/ha on level areas. Fertilizer rates are designed to provide some nutrient support to establishing vegetation, but not so much nutrient, especially N, that the treatment will promote the rapid establishment of weedy species such as foxtail barley.

As the 2017 summer season developed at Minto, it became clear that seeding would not take place in the fall of 2017 on any of these areas. Seed and fertilizer mixes are now on site and ready for application in 2018.

Projection of post-closure ecosites

A preliminary projection of post-closure ecosites on the reclamation-ready landforms was completed using information on substrate physical properties, cover depth and characteristics, topography, and ecological landscape classification systems developed by Environment Yukon. This work extends previously introduced concepts of plant-available water-storage capacity (AWSC) and soil moisture regime (SMR) for mine reclamation (Straker et al., 2015a, 2015b). Results from these projections are that post-closure ecosystems are projected to have mesic (zonal) or submesic (drier) SMRs. Examples of edaphic projections for both mesic and submesic sites are included as Appendix B.

We attempted to translate the SMR projections to the Yukon Ecological Landscape Classification system using documents produced by Environment Yukon (2017 and 2016). However, detailed description of ecosites associated with SMR positions has to date only been completed for the Boreal Low Southern Lakes (BOLsl) subzone, whereas the Minto mine footprint is located in the Boreal Low Yukon Plateau North (BOLyn) subzone. Given the lack of better information, and the relative proximity and similarity of the BOLsl and BOLyn subzones, we have at this time adapted information and the classification approach from the BOLsl guide (Environment Yukon 2017) to Minto, modified for local site observations, primarily those contained in a 2013 Access Consulting Group report.

The majority of the currently projected post-closure ecosystems are in the drier (submesic) 10 ecosite, while a smaller proportion (on tailings substrates and north-facing slopes) is in the zonal (mesic) 30 / 01 ecosite. We provide additional detail on these ecosites below.

- **Drier submesic 10 ecosite –** these ecosystems are generally lodgepole-pine- and trembling-aspen-leading forests, with smaller components of grasses and willows. Other commonly observed species include white spruce, balsam poplar, alders, kinnikinnick, mosses, and lichens.
- **Zonal submesic 30 / 01 ecosite –** these ecosystems are also pine- and aspen-leading forests with willow, but also have larger components of species associated with greater amounts of soil water, such as spruce and alder. Other commonly observed species include Alaskan birch, soapberry, *Vaccinium* and *Ledum* species, lupine, grasses, mosses, and lichens.

These ecosite projections will be used to develop revegetation treatments to supplement seeding as necessary. The primary reason to use these treatments will be to accelerate

revegetation, to establish species important for end land uses that are not establishing through natural regeneration, and/or for specific purposes such as additional erosion control around drainage channels. As discussed above, these projections and associated ecosite information are preliminary, being a transposition from the BOLsl ecosite classification, and based on sparse information on pre-development conditions at Minto. They will be refined as we continue to develop our projection system, as we acquire additional information on pre-development/adjacent (reference) ecosystems at Minto, and as the Yukon Ecological Landscape Classification system advances.

Closure

We trust this memo meets your information requirements at this time, and thank SRK for IEG's ongoing involvement with the Minto reclamation program. To discuss the contents of this memo and/or work required going forward, please do not hesitate to contact me at jstraker@iegconsulting.com, or at 250 701 0600.

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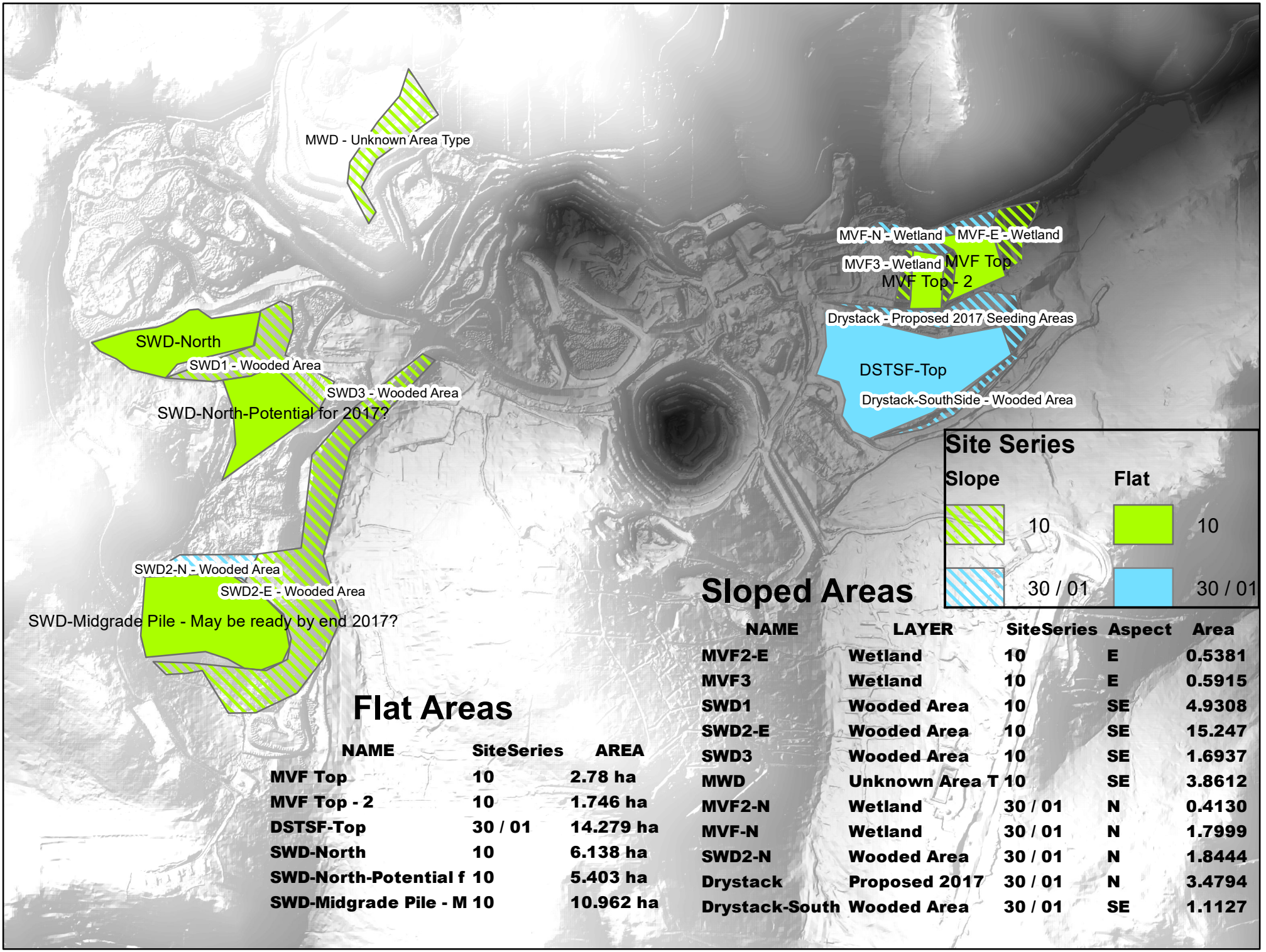
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Appendix A

Mapped revegetation units



Flat Areas

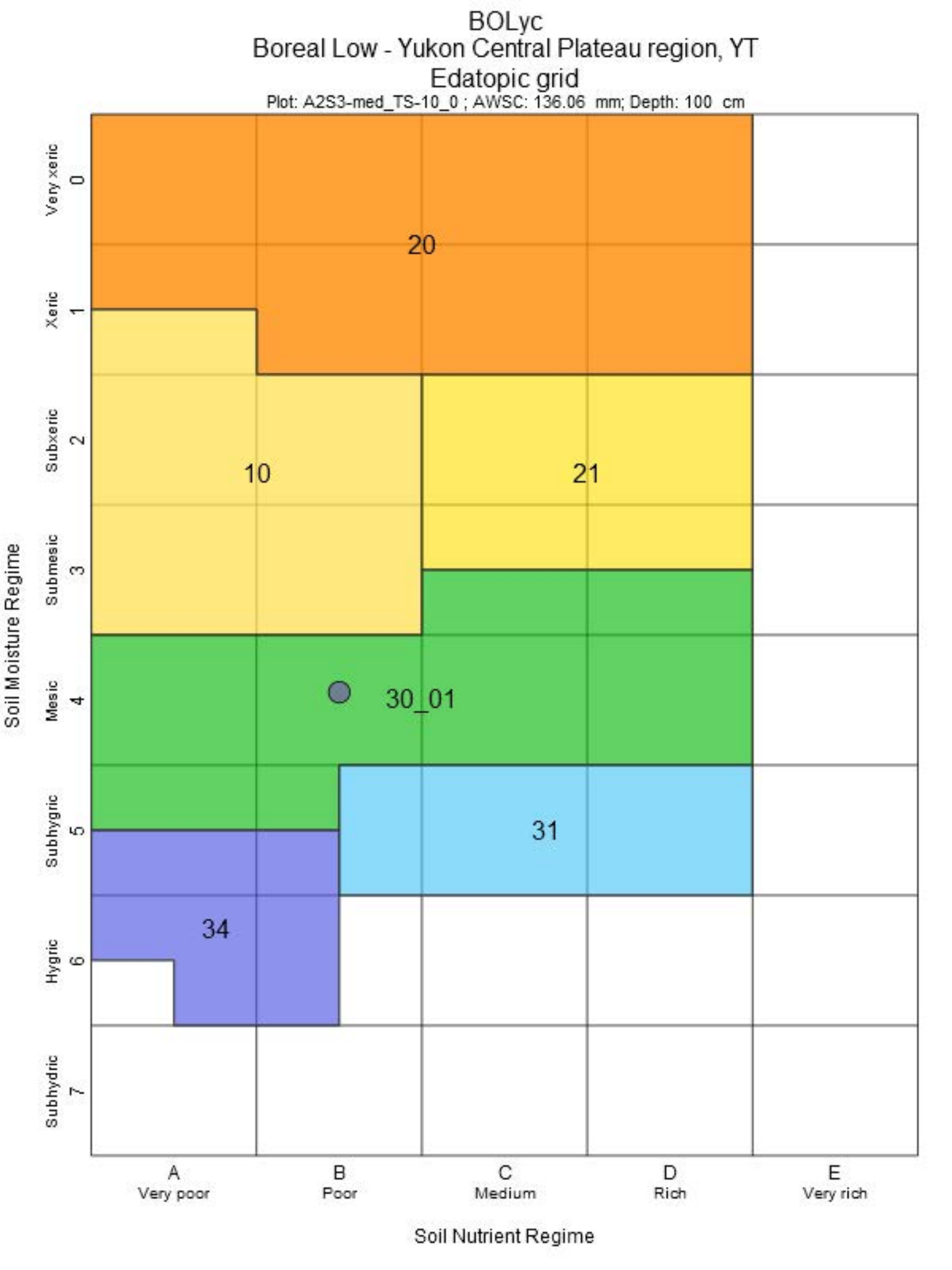
NAME	SiteSeries	AREA
MVF Top	10	2.78 ha
MVF Top - 2	10	1.746 ha
DSTSF-Top	30 / 01	14.279 ha
SWD-North	10	6.138 ha
SWD-North-Potential f	10	5.403 ha
SWD-Midgrade Pile - M	10	10.962 ha

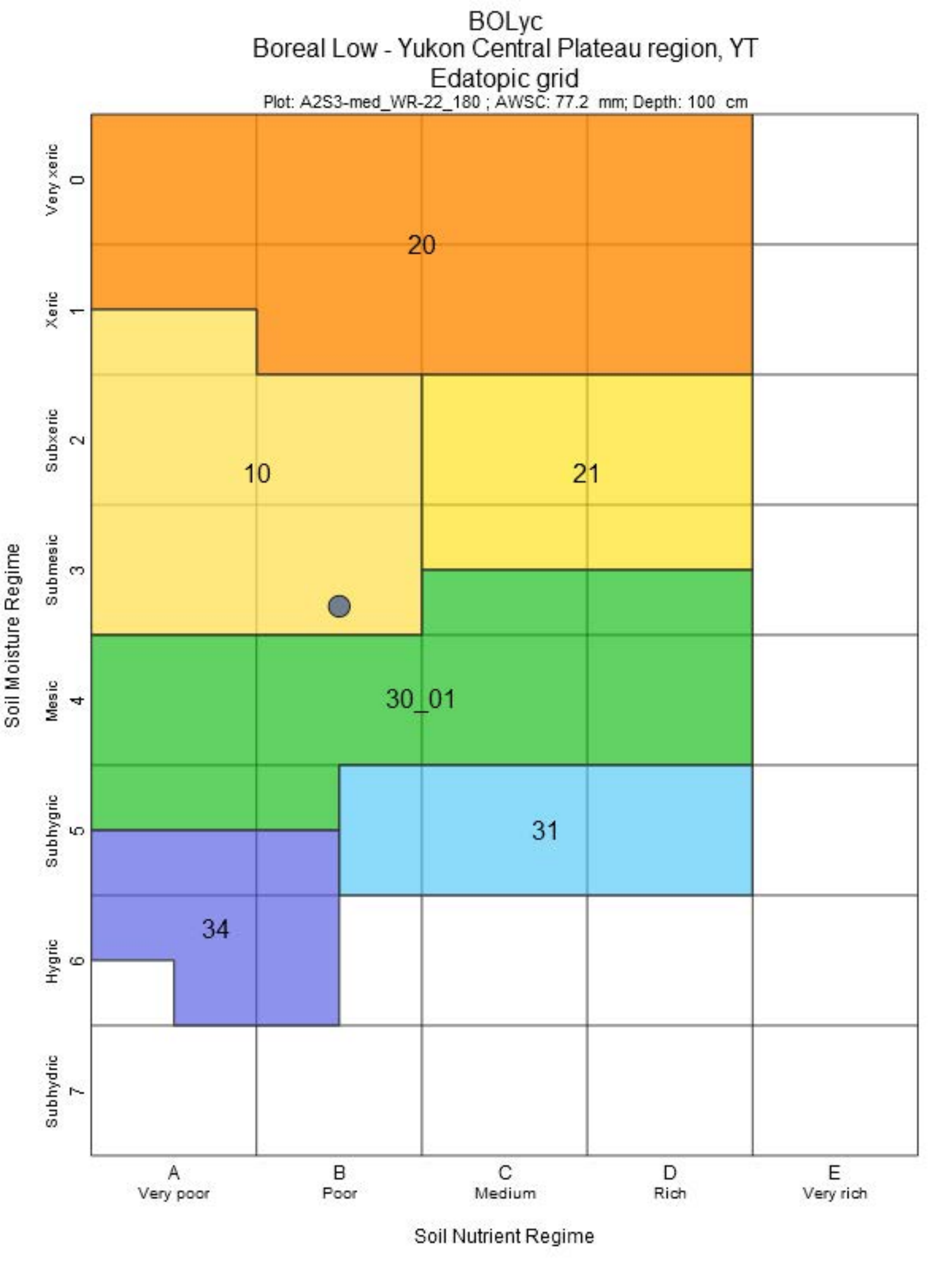
Sloped Areas

NAME	LAYER	SiteSeries	Aspect	Area
MVF2-E	Wetland	10	E	0.5381
MVF3	Wetland	10	E	0.5915
SWD1	Wooded Area	10	SE	4.9308
SWD2-E	Wooded Area	10	SE	15.247
SWD3	Wooded Area	10	SE	1.6937
MWD	Unknown Area T	10	SE	3.8612
MVF2-N	Wetland	30 / 01	N	0.4130
MVF-N	Wetland	30 / 01	N	1.7999
SWD2-N	Wooded Area	30 / 01	N	1.8444
Drystack	Proposed 2017	30 / 01	N	3.4794
Drystack-South	Wooded Area	30 / 01	SE	1.1127

Appendix B

Example edaphic-grid projections for mesic and submesic post-closure ecosites


BOLyc
Boreal Low - Yukon Central Plateau region, YT
Edatopic grid
Plot: A2S3-med_TS-10_0 ; AWSC: 136.06 mm; Depth: 100 cm
Soil Moisture Regime
0 Very xeric
1 Xeric
2 Subxeric
3 Submesic
4 Mesic
5 Subhygric
6 Hygric
7 Subhydric
20
10
21
30_01
31
34
A Very poor
B Poor
C Medium
D Rich
E Very rich
Soil Nutrient Regime


BOLyc
Boreal Low - Yukon Central Plateau region, YT
Edatopic grid
Plot: A2S3-med_WR-22_180 ; AWSC: 77.2 mm; Depth: 100 cm
Soil Moisture Regime
0 Very xeric
1 Xeric
2 Subxeric
3 Submesic
4 Mesic
5 Subhygric
6 Hygric
7 Subhydric
20
10
21
30_01
31
34
A Very poor
B Poor
C Medium
D Rich
E Very rich
Soil Nutrient Regime

Appendix F – Minto Closure Cover Design – Hydrotechnical Designs for Engineered Landforms

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Memo

To:	File	**Client:**	Minto Explorations Ltd.
From:	Erik Ketilson, MEng, PEng	**Project No:**	1CM002.049
Reviewed by:	Dylan MacGregor, MSc, PGeo	**Date:**	January 29, 2018
Subject:	Minto Closure Cover Design – Hydrotechnical Designs for Engineered Landforms		

1 Introduction

SRK Consulting (Canada) Inc. (SRK) is undertaking an update to the Minto Closure Plan, which includes updated landform grading designs for the waste facilities, and consideration to shedding water off of these structures. Several areas of the site require varying degrees of hydrotechnical features to collect and convey storm water over the covers. Wide, shallow channels (swales) have been sized for the main engineered landforms based on catchment area and grade. The main engineered landforms under consideration include the Southwest Waste Dump (SWD), the Ridgetop Dump (RTD), the Rigetop North Pit Tailings Management Facility (RNPTMF), the Ridgetop South Pit Backfill Dump (RSPBD), the Dry Stack Tailings Storage Facility (DSTSF), the Mill Valley Fill (MVF), the Main Waste Dump and Main Waste Dump and Expansion (MWD), the Main Pit Dump (MPD), and the Area 118 Pit Backfill Dump (Area 118). This memo summarizes the design process and presents the recommended sizing of the swale designs for each area.

2 Watershed Analysis

Watershed areas were delineated for each of the eight main landforms based on closure configurations as presented in the Updated Closure Cover Design Report (SRK, 2018). Each landform, with the exception of Area 118, was then subdivided into smaller watershed areas that will collect water independently of one another to shed water off the top of the structure, and then into a channel directed down-slope. The areas and their general flow directions are presented in Figure 1.

3 Hydrology

Site Hydrology was updated by SRK in 2016; complete hydrological details are included in The Minto Mine – Closure Water Conveyance System Design Update Report (SRK, 2016). The updated flow vs. watershed area relationships yield substantially greater flows than in previous site hydrological assessments completed by SRK and Janowicz (SRK, 2016). The increase is due in part to the fact that for return periods greater than 1:5 year event, Environment Canada's hydrometric station, the Little South Klondike River below Ross Creek presents the highest unit

peak. For this reason, the Little South Klondike River below Ross Creek hydrometric station was selected for calibration purposes in order to generate a conservatively high estimate for unit peak flows.

The swales were designed to accommodate a 1 in 200 year, 24 hour storm event consistent with the approach presented by SRK (2016). The flows expected on each watershed are presented in Table 3-1, with watershed areas illustrated on Figure 1.

Table 3-1: Watershed Areas and Design Flow Rates

Area	ID	Contributing Area (m^2)	1 in 200 Year Flow Rate (m^3/s)
SWD	A1	11,500	0.09
	A2	11,170	0.09
	A3	20,830	0.15
	A4	18,840	0.13
	A5	12,050	0.09
	A6	17,360	0.13
	A7	111,110	0.63
	A8	19,540	0.14
	A9	36,800	0.24
	A10	32,960	0.22
	A11	59,870	0.37
RTD	A12	34,820	0.23
	A13	28,040	0.19
	A14	7,380	0.06
DSTSF	A15	30,490	0.20
	A16	22,100	0.15
	A17	20,540	0.15
	A18	63,160	0.39
	A19	15,310	0.11
MVF	A20	32,450	0.22
	A21	120,380	0.68
	A22	55,470	0.34
MWD	A23	49,180	0.31
	A24	27,240	0.19
	A25	78,600	0.47
	A26	102,770	0.59
MP	A27	26,120	0.18
A118	A28	22,670	0.16
	A29	31,780	0.21
	A30	7,790	0.06
RNPTMF	A31	110,600	0.63
RSPBD	A32	27,040	0.18

Source: 2018-Minto Channel Flow_Rev03-jg-ek.xlsx

4 Swale Designs

Drainage swales have been designed for the top surfaces and slopes of the seven main engineered landforms. The channel depths were determined using Manning's Equation, and were then increased by a factor of 2 to account for the potential effects that ice may have on flow within the channel. This approach is consistent with other sites in the Yukon, where there is a need to account for increased channel capacity and quantification of the increased capacity is challenging.

The cross sections of the swales on the top surfaces are trapezoidal, primarily with a base width of 2 m and 10H:1V side slopes. Each swale was designed with a variable grade, representative of the anticipated grade of the final closure cover design, the grades are presented in Table 4-1. A Manning's "n" in each channel of 0.035 was applied, which is representative of roughness due to the presence scattered brush and heavy weeds in flood plains (Bedient et al, 2008).

The cross sections of the swales on the slopes are trapezoidal, with base widths equal to those of the swales on the top surfaces, but with 3H:1V side slopes. Each swale was designed with a variable grade, representative of the closure cover design, the grades are presented in Table 4-2. A Manning's "n" in each channel of 0.040 was applied, which is representative of roughness due to the presence a bottom with gravels, and cobbles in mountain streams (Bedient et al, 2008).

The dimensions of each channel are indicated in Table 4-1 for the facility tops, and Table 4-2 the slopes. Flow velocity is also included in the table.

Table 4-1: Channel Dimensions and Flow Velocities for Facility Tops

Area	ID	Channel Dimensions				Channel Dimensions for n=0.035 - vegetation				
		Slope (%)	Length (m)	Bottom Width (m)	Side Slopes (H:V)	Depth of Flow (m)	Allotment for Ice Formation (m)	Total Depth (m)	Top Width (m)	Flow Velocity (m/s)
SWD	A1	1.5%	100	2.0	10	0.07	0.07	0.13	4.7	0.49
	A2	1.5%	85	2.0	10	0.07	0.07	0.13	4.6	0.49
	A3	2.2%	110	2.0	10	0.08	0.08	0.16	5.2	0.66
	A4	2.0%	110	2.0	10	0.08	0.08	0.16	5.1	0.62
	A5	1.0%	85	2.0	10	0.08	0.08	0.15	5.0	0.44
	A6	1.0%	110	2.0	10	0.09	0.09	0.18	5.6	0.48
	A7	1.5%	85	2.0	10	0.19	0.19	0.37	9.5	0.88
	A8	1.5%	115	2.0	10	0.09	0.09	0.17	5.4	0.57
	A9	0.5%	140	2.0	10	0.15	0.15	0.30	8.1	0.45
	A10	4.5%	140	2.0	10	0.08	0.08	0.16	5.3	0.96
	A11	2.0%	300	2.0	10	0.13	0.13	0.26	7.3	0.84
RTD	A12	0.2%	300	2.0	10	0.19	0.19	0.37	9.4	0.32
	A13	2.5%	120	2.0	10	0.09	0.09	0.18	5.5	0.75
	A14	0.2%	85	2.0	10	0.09	0.09	0.19	5.7	0.22
DSTSF	A15	1.5%	140	2.0	10	0.11	0.11	0.21	6.2	0.64
	A16	1.5%	140	2.0	10	0.09	0.09	0.18	5.6	0.59
	A17	2.5%	250	2.0	10	0.08	0.08	0.15	5.0	0.69
	A18	2.0%	225	2.0	10	0.14	0.14	0.27	7.4	0.85
	A19	3.5%	105	2.0	10	0.06	0.06	0.12	4.4	0.72
MVF	A20	2.5%	200	2.0	10	0.09	0.09	0.19	5.8	0.78
	A21	1.5%	165	2.0	10	0.19	0.19	0.39	9.7	0.89
	A22	1.5%	125	2.0	10	0.14	0.14	0.28	7.5	0.74
MWD	A23	1.5%	170	2.0	10	0.13	0.13	0.26	7.2	0.72
	A24	5.5%	175	2.0	10	0.07	0.07	0.14	4.7	0.98
	A25	2.5%	180	2.0	10	0.14	0.14	0.28	7.6	0.97
	A26	2.0%	125	2.0	10	0.17	0.17	0.34	8.7	0.95
MP	A27	1.0%	150	2.0	10	0.11	0.11	0.22	6.4	0.53
A118	A28	3.0%	65	2.0	10	0.08	0.08	0.15	5.0	0.75
	A29	5.0%	50	2.0	10	0.08	0.08	0.16	5.1	0.98
	A30	5.0%	25	2.0	10	0.04	0.04	0.08	3.6	0.67
RNPTMF	A31	0.5%	140	2.0	10	0.24	0.24	0.48	11.7	0.59
RSPBD	A32	1.0%	80	2.0	10	0.11	0.11	0.22	6.4	0.54

Source: 2018-Minto Channel Flow_Rev03-jg-ek.xlsx

Table 4-2: Channel Dimensions and Flow Velocities for Facility Slopes

Area	ID	Channel Dimensions				Channel Dimensions for n=0.040 - gravel to cobble sized rip rap				
		Slope (%)	Length (m)	Bottom Width (m)	Side Slopes (H:V)	Depth of Flow (m)	Allotment for Ice Formation (m)	Total Depth (m)	Top Width (m)	Flow Velocity (m/s)
SWD	A1	10.5%	130	2.0	3	0.04	0.04	0.04	2.3	0.95
	A2	21.0%	85	2.0	3	0.03	0.03	0.07	2.4	1.17
	A3	19.5%	100	2.0	3	0.05	0.05	0.10	2.6	1.40
	A4	10.0%	350	2.0	3	0.06	0.06	0.11	2.7	1.10
	A5	19.0%	80	2.0	3	0.04	0.04	0.07	2.4	1.17
	A6	12.0%	110	2.0	3	0.05	0.05	0.10	2.6	1.14
	A7	22.0%	150	2.0	3	0.11	0.11	0.22	3.3	2.45
	A8	19.0%	185	2.0	3	0.05	0.05	0.10	2.6	1.36
	A9	10.5%	284	2.0	3	0.08	0.08	0.16	2.9	1.38
	A10	8.3%	100	2.0	3	0.08	0.08	0.16	3.0	1.23
	A11	20.0%	115	2.0	3	0.08	0.08	0.17	3.0	1.97
RTD	A12			NA				NA		
	A13			NA				NA		
	A14			NA				NA		
DSTSF	A15			NA				NA		
	A16			NA				NA		
	A17	17.0%	65	2.0	3	0.10	0.10	0.21	3.2	2.03
	A18									
	A19			NA				NA		
MVF	A20			NA				NA		
	A21	25.0%	55	2.0	3	0.11	0.11	0.22	3.3	2.62
	A22	30.0%	40	2.0	3	0.07	0.07	0.14	2.9	2.19
MWD	A23	30.0%	35	2.0	3	0.07	0.07	0.13	2.8	2.11
	A24	33.0%	40	2.0	3	0.05	0.05	0.10	2.6	1.81
	A25	22.5%	75	2.0	3	0.14	0.14	0.28	3.7	2.85
	A26									
MP	A27	42.0%	35	2.0	3	0.04	0.04	0.09	2.5	1.92
A118	A28	40.0%	150	2.0	3	0.04	0.04	0.08	2.5	1.81
	A29	35.0%	200	2.0	3	0.05	0.05	0.10	2.6	1.93
	A30	10.0%	200	2.0	3	0.04	0.04	0.07	2.4	0.83
RNPTMF	A31			NA				NA		
RSPBD	A32	10.0%	65	2.0	3	0.07	0.07	0.14	2.8	1.23

Source: 2018-Minto Channel Flow_Rev03-jg-ek.xlsx

5 Conclusion

As stated by SRK (2016), engineered designs to reduce erosion within the swales may be required for flow velocities greater than 1 m/s (engineered designs could include rip rap, other armouring, or velocity reducing features within the swale). Engineered designs to reduce erosion in cases where flow velocities are less than 1 m/s are not required; in these scenarios, short term erosion control support practices, such as the use of rolled erosion control products and/or hydroseeding is sufficient to reduce erosion until surface vegetation is established.

The swales for the facility tops have a base width of 2 m and are 0.08 to 0.48 m deep and have 3.6 to 11.7 m top widths. All designs were completed to equal a flow velocity of 1.0 m/s or less such that channel protection can be achieved using surface vegetation.

The swales for the facility slopes have base widths equal to the facility tops. Total depths were estimated to be between 0.04 m and 0.28 m deep, and have top widths ranging between 2.26 m and 3.68 m. Nearly all of the flow velocities are greater than 1.0 m/s, while considering a base consisting of gravel and cobble rip rap protection. Therefore, detailed design of these structures should account for appropriate riprap protection, with sizing of the rip rap to be completed at that time.

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The opinions expressed in this report have been based on the information available to SRK at the time of preparation. SRK has exercised all due care in reviewing information supplied by others for use on this project. Whilst SRK has compared key supplied data with expected values, the accuracy of the results and conclusions from the review are entirely reliant on the accuracy and completeness of the supplied data. SRK does not accept responsibility for any errors or omissions in the supplied information, except to the extent that SRK was hired to verify the data.

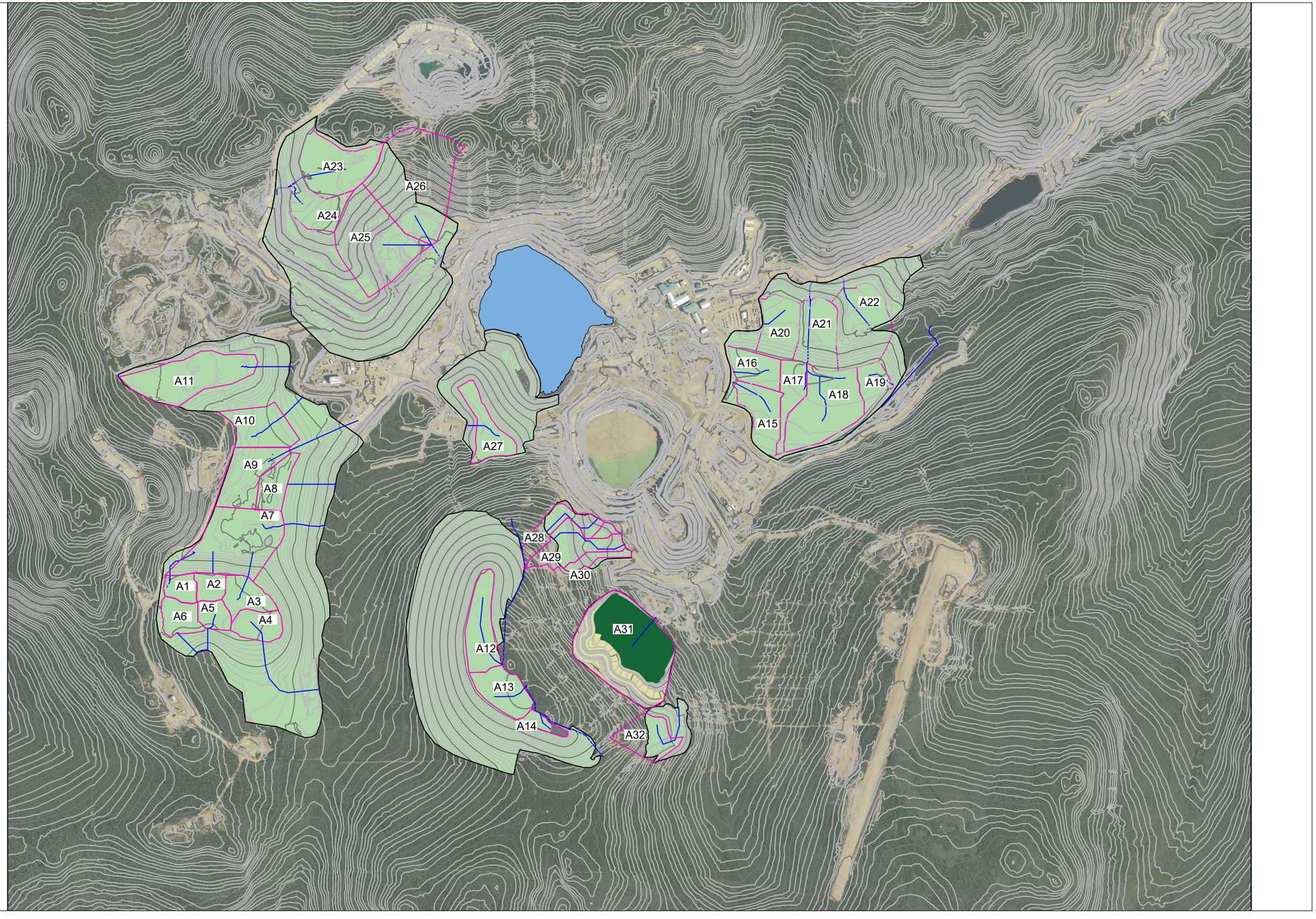
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Figures



LEGEND

- Watershed
- A12 Watershed Reference Number
- Proposed Flow Alignment
- Subaerial Tailings Cover
- Cover Application Area
- Open Pit Slopes
- Water

NOTES

1. Contours shown at 2.0m intervals.
2. Orthophoto date August, 2017 and provided by Minto.
3. This figure should be read in conjunction with report Minto Closure Cover Design - Hydrotechnical Designs for Engineered Landform. Dated January, 2018.
4. Waste facility grading is based on designs prepared for the Updated Closure Cover Design for the Minto Mine 2018 Reclamation and Closure Plan. Dated January, 2018.

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srk consulting

SRK Project NO.: 1CM002.049

FILE NAME: 1CM002.049_2018RCP-Regrade-Catchments_R1.dwg

capstone MINTO MINE
OPERATED BY MINTO EXPLORATIONS LTD.

Minto Mine, Yukon

Hydrotechnical Designs for Covers

Watershed Areas

DATE:	APPROVED:	FIGURE:
January 2018	EK	01

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Closure Landform Design and Reclamation Landform Units for the Minto Mine

Prepared for

Minto Explorations Ltd.



Prepared by

SRK Consulting (Canada) Inc.
1CM002.031
May 2016

Closure Landform Design and Reclamation Landform Units for the Minto Mine

May 2016

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1 Introduction

1.1 Site Description

The Minto Mine is a high-grade copper mine located in the Yukon, approximately 240 km north of Whitehorse. The mine site occupies the valley in the upper reaches of Minto Creek, a tributary on the west side of the Yukon River, about 9 km from the mouth. Operations are ongoing at this time (2016) and began in October 2007. Three pits have been completed to date: the Main Pit, the Area 118 Pit, and the Area 2 Stage 2 Pit.

Initially tailings were deposited in the Dry Stack Tailings Storage Facility (DSTSF) which was completed in 2012. Currently tailings and process water are managed in the Main Pit Tailings Facility (MPTF) and the Area 2 Pit Tailings Management Facility (A2PTMF). Various waste rock storage facilities exist across the site including the Main Waste Dump (MWD), Main Waste Dump Expansion (MWDE), the Main Pit Dump (MPD), the Southwest Waste Dump (SWD), the Area 118 Dump, and the Mill Valley Fill (MVF) currently undergoing the Stage 2 expansion.

1.2 Purpose of this Report

Some mine structures are complete (e.g. DSTSF) or are nearing completion (e.g. Main Waste Dump Expansion) and are in need of reclamation designs. In order to facilitate the reclamation of various structures across the site, Minto Explorations Ltd. (Minto) has retained SRK Consulting (Canada) Inc. (SRK) to prepare general landform design concepts specific to the Minto site, and to define Reclamation Land Units (RLUs) that will serve as the building blocks for establishing closure landforms across the site.

This report documents the work completed to develop site-specific landform design concepts and RLU designations for the Minto Mine.

1.3 Scope of Work

As part of the larger project scope, site reconnaissance was completed in 2014. Based on the information gleaned from the site visit, coupled with other relevant information such as closure cover soil availability and type, SRK has prepared general landform design concepts or “rules” specific to the overall Minto site. The purpose is to develop landform design concepts that will ensure geotechnical stability and landforms that have a natural appearance that fits with the surrounding landscape, and that drains surface water, while minimizing the potential for erosion. Reclamation Land Units (RLU) will also be developed for the site. Each RLU will have specific function(s) and will be used in future design stages as the guiding principles for establishing a closure landform.

This report documents the landform design concepts and RLUs and is the deliverable for this task. Some cost considerations are also provided, together with landform monitoring recommendations.

It is intended that this work will become the basis for all future site reclamation and development planning. This document should be considered a “live” document, as it is expected to be adjusted as progressive reclamation takes place and improvements are made based on lessons learned along the way.

1.4 About this Report

This Report is organised into seven sections, excluding the references section. Following the introduction (Section 1), relevant site background description is provided in Section 2. The technical analysis part of the report is comprised of three sections, with Section 3 presenting the soil erosion analysis completed, Section 4 providing the methodology and results of the conceptual landform design while Section 5 presents a description of the RLU designations. The erosion analysis and the landform design process are equally important in informing the RLU descriptions, which in turn will become an important part of final reclamation planning for Minto Mine. Section 6 discusses the RLU designations and their application to closure planning at Minto.

2 Site Background

The general conditions at the Minto Mine site are discussed in a number of public documents related to permitting, operation, expansion and closure of the mine (Access 2013, Access 2014, SRK 2014). This section of the report will therefore focus primarily on those topics which are of direct relevance.

2.1 Overburden

Overburden thickness across the site is closely correlated with geomorphological features. Near topographic highs, there is little to no overburden, with increasing overburden thickness down valley slopes. The thickest deposits are found in the valley bottoms and along the paleochannel, which is offset to the south from the Minto Creek valley bottom and identified on the south side of the Main Pit and north side of the DSTSF (SRK 2014).

Generally, ridge tops are dominated by sandy residual soils grading to weathered bedrock. Fine weathered products have been washed downslope, and have also been deposited by glacial, periglacial and aeolian processes, which has resulted in accumulations of finer sandy silts and clays within the valleys.

2.2 Permafrost

Minto Mine is located in a discontinuous permafrost region. In general, the south facing slopes are unfrozen, while north facing slopes and valley bottoms contain permafrost. Where permafrost is present it is generally warm, with temperatures close to -0.5^{o}C. The active layer is variable ranging from less than 1 m to about 3 m in thickness (SRK 2014). The general distribution of permafrost on the site is illustrated in Figure 1.

Ice content within the overburden ranges from non-visible to 30% excess ice, although ice lenses are common and massive ice up to 4 m thick occurs, particularly near the bedrock surface within the paleochannel. The depth of permafrost is variable, but in certain locations it is known to extend down to the base of the paleochannel.

2.3 Vegetation

Vegetation on site is dominated by mixed Willows and Trembling Aspen, with fire history (and related vegetative succession) exerting a very strong influence on the present distribution of plant species. Lodgepole Pine is a later successional species that will dominate mid and upper well drained slopes. White and Black Spruce seedlings are common in the undergrowth and will likely outgrow the pine and aspen communities, especially on northern aspects and along slope toes and valley bottoms. Small grassland areas exist along ridge-tops and steep south-facing slopes as these areas do not retain enough moisture to support tree growth. Much of the planned mine expansion is along ridgetops and mid-slopes, with vegetation in these areas being mostly comprised of willow, aspen, pine, and understory seedling growth. No rare or endangered plants were found during surveys of the planned impact areas (Access 2013).

It should be noted that vegetation trials were conducted at Minto until 2015. Vegetation test plots on the sloped face of the Main Waste Dump were being actively managed and assessed annually and findings from this trial program will be transferred to more progressive reclamation efforts at other areas of the site. As trial results become available, knowledge relating to growth media, seed mixes, vegetation establishment, and vegetative erosion control will evolve and reclamation land unit designations will be revised accordingly (Access 2014).

3 Erosion Analysis

Land impacted by mining is often characterized by steep slopes as illustrated in Figure 2. Waste rock dumps are usually constructed by end-dumping from slope crests such that the dump material lies at its natural angle of repose. Dump faces constructed in this manner often have a high erosion potential due to the uniformly steep and relatively long slopes. In cases where soil covers are required to achieve the long-term closure objectives, the slope of these dump faces needs to be reduced to allow the construction of a soil cover.

Even soil covers constructed on regraded slopes (typically much shallower than angle of repose) may exhibit excessive erosion that could, under certain conditions, compromise the overall performance. It is therefore important to identify if erosion could become a failure mechanism. This is typically done through erosion modelling, as described below.

3.1 Erosion Assessment Methodology

Various numerical methods are in use worldwide to determine soil loss on an inclined slope, most originating from agriculture and soil sciences. One of the most commonly used models in North America is the USLE model, short for Universal Soil Loss Equation, which was developed using runoff and soil loss data from over 10,000 plot years at 49 locations across the US in the 1950s (USDA 1978). The USLE was developed to predict the long term average annual rate of erosion

on a sloped field and is based on a combination of rainfall pattern, soil type, topography, crop type, and management practices. One of the major limitations of this approach is that it assumes sheet erosion as the only erosional mechanism, whereas rill erosion is typically predominant for recently placed soil cover material on slopes.

Although rudimentary in its approach, the USLE method was chosen for its simplicity, as the intent here was to demonstrate relative sensitivity of slope erosion to the various environmental factors affecting the slope, rather than compute a highly accurate soil loss estimate. Using a rigorous erosion prediction method was necessary to compare the various slope configurations and identify the range in which each input parameter will exert a strong influence on performance.

The erosion assessment approach may be re-evaluated at a later stage (with consideration given to using a more sophisticated model), when specific slope designs will be completed for the distinct mine waste facilities.

An estimate of the potential erosion from slopes with a soil cover at gradients ranging from 2.5H:1V to 6H:1V was completed using the USLE. At the Minto site, conservation measures similar to those used in farm planning can be applied to evaluate closure options among landform designs, and were therefore considered in the erosion estimates.

The governing equation is presented below:

$$A = R * K * LS * C * P \qquad \text{[Eq. 1]}$$

Where;
R = rainfall and runoff factor (dimensionless)
K = soil erodibility factor ($\mathrm{t/Ha}$*)*
LS = slope length and steepness factor (dimensionless)
C = crop/vegetation management factor (dimensionless)
P = support practice factor (dimensionless)

The slope length and steepness factor (LS) is calculated using the following equation:

$$LS = (0.065 + 0.0456S + 0.006541S^2) * (L/22.1)^X$$

Where:
S = slope steepness (%)
L = slope length (m)
X = 0.2 (S<1%), 0.3 (1≤S<3), 0.4 (3≤S<5), 0.5 (≥5)

3.2 USLE Parameter Selection

Table 1, at the end of this section, summarizes the parameters used in the USLE calculation. For the rainfall and runoff factor, it was decided to use a rainfall and runoff factor from an area in the United States of America with a similar mean annual precipitation (MAP). The Minto project has a MAP of approximately 350 mm. Areas in the mid-west US with comparable MAP have rainfall and runoff factors of 75 or less. It has been assumed that a value of 75 is appropriate for the Minto site.

For the soil erodibility factor, it was assumed that cover material will be sourced from the reclamation overburden dump (ROD) or from stripping related to the Phase V/VI pit development activities. On average, the ROD material has an average of 17% gravel, 46% sand, 27% silt, and 10% clay (SRK 2013). Removing the gravel proportion, this equates to normalized sand, silt, and clay percentages of 55, 33, and 12, respectively, which means the fine portion of the material falls into the Sandy Loam soil category and is on the fine end (Figure 3 (a)). Several sources provide soil erodibility factors for loam soils that fit this general soil type. A factor of 0.54 was chosen as the average between sandy loam (0.31) and loam (0.71) (Stone and Hilborn 2015). However, the factor was varied for different cover soil types as illustrated in Figure 3 (b).

Slope-length and slope steepness (gradient) were varied in the calculations to allow a comparison of the calculated erosion rates at different slope lengths and gradients. Slope lengths and gradients will vary on reclaimed landforms as they vary across the unimpacted areas of the site, so the analysis focused on slopes between 22° (2.5H:1V) and 9° (6H:1V) with lengths between 20 m and 200 m. Figures 3 (c) and (d) illustrate the change in calculated erosion values for this range of slope lengths and gradients.

The crop and vegetation factor is a reduction factor based on the type of erosion protection given by particular vegetation types, ranging from 0.01 for anti-erosion blankets and mats to 1.0 for no vegetation. For mine reclamation it is important to understand the relationship between erosion and different levels of vegetation establishment. Figure 3 (e) includes an analysis of erosion based on vegetation factors covering the entire available USLE model input range, from no vegetation to purpose-built anti-erosion mats. This range also includes trees, bushes, straw mulch, and grass.

The support practice factor is included to characterize the effect of particular cover surface treatment. A smooth surface will increase erosion as it allows surface water to move with maximum energy, while tracked surfaces provide natural kinetic energy reduction due to micro-scale step pools that help dissipate energy. The support practice factor for agricultural practices like cross-slope farming and contour farming (analogues for cross-slope tracking or ripping) ranges between 0.25 and 1.0. However, since mine reclamation does not typically involve annual re-work of surfaces, as is common with farm applications, it was assumed that no improvements will be maintained in the long term and the support practice factor has been set to 1.0 for all analyses.

Table 1: Parameters Selected in USLE Calculation

USLE Factor	Base Case	Variation
Rainfall and Runoff Factor – R (dimensionless)	75	N/A
Soil Erodibility Factor – K (t/ha)	0.54	0.07 to 0.74 for different soil types.
Slope Length and Steepness – LS (dimensionless)	18.83 (100 m slope length and 3:1 slope)	5.62 to 26.63 for slope lengths between 20 and 200 m and slopes from 2.5:1 to 6:1.
Crop/Vegetation Management Factor – C (dimensionless)	0.25 (no cover and no tillage)	0.005 to 0.25 for different vegetation/cover treatments from erosion blankets to trees to no cover. Tillage not considered in range.
Support Practice Factor – P (dimensionless)	1	N/A

3.3 USLE Analysis Results

A base case for erosion at the Minto site was defined in order to allow comparison of erosion rates for differing soil types, vegetation, and slope lengths and gradients. Erosion for the base case was calculated for a sandy loam on a slope 100 m long at 18° (3H:1V) with no vegetation or support management. The base case estimated an annual erosion of 270 tonnes/hectare, which is considered severe in terms of crop soil loss. Single parameter values were then varied in the equation from the base case in order to illustrate the influence of each parameter on the final landform performance. Comparative graphs of the erosion estimates are shown in Figure 3 (b) through (e).

The analysis showed that erosion can be minimized by reducing slope lengths and gradients, using sandy covers, and ensuring thorough revegetation. The relationship between the amount of erosion and slope length is clear and almost linear – as slope length increases, so does erosion. As slope gradient decreases, erosion decreases; however, the effect becomes less pronounced at slope gradients shallower than 11° (5H:1V). This is important to note as regrading large, steep slopes to long, flat ones can be costly. From this analysis, it appears that short slopes covered with sand and densely vegetated with an angle of 11° or less will yield the least erosion from final landforms at Minto. This analysis shows that flatter slopes lead to reduced erosion.

In practical terms, slope lengths can be reduced by grading landforms such that multiple micro-catchments are constructed within a single landform. If absolutely necessary, sub-horizontal benches that reduce the cumulative energy of surface water can be constructed in such a way that standing water is avoided but that flow along the landform is still slowed. Revegetation also plays a large role in minimizing erosion as shown in Figure 3 (e), so landforms should be revegetated with appropriate mixes as soon as possible after landform construction has been completed.

4 Landform Design

4.1 Methodology

4.1.1 Landform Design Concepts

Landform design which adheres to the principles of fluvial geomorphology aims to emulate the characteristics of mature, natural landforms with fully developed and self-healing drainage systems. Such landforms are capable of adapting to geomorphological changes without accelerated erosion or unacceptable environmental impacts (Sawatsky and Beckstead 1995; Toy and Chuse 2005). Characteristics of landforms designed using these principles include:

- Slopes in equilibrium with local rainfall and soil conditions;
- Considerable variability in topographic relief;
- Well-defined watersheds with easily recognizable watercourses set in incised swales;
- Overland flow path lengths that do not violate thresholds that can be determined from local natural terrain;
- Watershed areas that do not exceed a threshold defined by the slope of their watercourses;
- Drainage systems patterned after natural streams in the surrounding area;
- Erosion rates comparable to those of the natural environment;
- Convex crests of increasing steepness transitioning to concave slope profiles that begin steep and become shallower along the flow path; and
- Drainage systems that are able to evolve and are “self-healing”.

Landforms designed using geomorphological principles have been shown to experience reduced erosion and improved hydrologic response to large storm events, as peak runoff rates and total runoff from these landforms have been shown to be lower than for traditionally engineered designs (Snyder 2013). Geomorphic landform designs have also been shown to have improved long term slope stability as there is often reduced migration of fine particles towards slope toes. This can improve slope stability by reducing the likelihood of slope toes becoming saturated (Russell 2012). The complex topography of geomorphic landforms has also been shown to increase flora diversity, which can lead to decreased erosion, improved slope stability, and better closure outcomes (Bugosh and Epp 2014). Figure 4 illustrates some of the landform design concepts described above.

4.1.2 Design Objectives

Closure landform design at the Minto Mine has the following objectives:

- Ensure the geotechnical stability of final landforms against surficial failures;
- Ensure the hydrotechnical stability of final landforms;

- Create conditions for the natural succession of vegetation on final landforms;
- Employ a combination of design slopes and vegetation to assist in maximizing erosional stability; and
- Develop criteria to assist with performance monitoring and maintenance of future reclaimed structures at the Minto Mine.

4.1.3 Landform Design Approach

In order to design landforms that meet the closure objectives, a methodology should be established that allows for investigation of natural landforms, site conditions, and possible techniques for incorporating natural landform features into final mine closure landforms. The following methodology is proposed:

- Observe natural landforms that are preferably mature and stable and note common features of these landforms, such as soil cover and moisture, slope length, profile, aspect and gradient, and vegetation cover;
- Look at site layouts and decide the course of action to create closure landforms that best emulate natural, stable landforms;
- Develop specific landform resloping designs that emulate natural landforms or the pre-mining landscape as much as practical;
- Evaluate options to create micro-landforms within larger landforms should it not be practical to emulate the natural landscape for large landforms; and
- Develop a progressive reclamation monitoring plan in order to monitor the performance of the designed landforms.

The above methodology was developed based on the principles of geomorphic landform design and was followed in the design of the SWD regrading and progressive reclamation plan. It should be considered in all future reclamation designs at the Minto site in order to be consistent with the principles of geomorphic landform design.

4.2 Results of Site-Specific Landform Analysis

4.2.1 Site Reconnaissance

Two SRK engineers travelled to the Minto Mine site for three days in late May 2014 to observe the natural geomorphological features in the general area of the mine as well as the immediate vicinity of the various waste dumps and other facilities. The current shape of the SWD, MWD, and DSTF were also observed and relevant features noted. Specifically, on built landforms, engineers looked for geotechnical stability indicators, erosional features, vegetation, slope angles and aspects, particle sizes and type, water conveyance structures, and areas with ponded water. In general, it was noted that:

- Most built landforms appeared geotechnically stable, even when apparently over steepened slopes are visible, at or exceeding theoretical angles of repose. There were a number of small surficial skin failures evident on some of the finer grained waste rock dump slopes;
- Erosion was very visible on most built landforms, and ranged from sheet erosion to considerable progressive rill and gully erosion. Most of the progressive rill and gulley erosion appeared to be associated with long continuous runoff surfaces, or areas where water was allowed to concentrate. Interestingly, some of the flatter regraded slopes appeared to show more demonstrated evidence of erosion;
- Physical composition of the different waste and overburden materials are highly variable, ranging from well-graded to poorly graded compositions with particle sizes ranging from boulders to silt. There was however no clear evidence of substantial self-armoring of erosion gulley's in any of these materials. Many of the finer grained materials do exhibit measurable evidence of erosional deposition at the slope toes; and
- Some areas are revegetating naturally, but by far the predominant reason for natural revegetation appears to be in areas where organic rich overburden soils are stockpiled; Areas where active revegetation has been done on organic poor overburden soils, vegetation succession appears to be poor.

In natural undisturbed landscapes in the vicinity of the site, engineers noted slope profiles and aspects, vegetation type, the presence or absence of standing water, erosional features, and channel profiles, cross-sections, and armouring material. These aspects of the surrounding natural landforms were observed and the information gathered will be used to inform the design of landforms that mimic the local natural geomorphic features. In general, it was noted during the site visit that natural landforms tended to have relatively shallow slopes with limited evidence of erosion. There was however clear evidence of well-developed gulley's parallel to most slopes suggesting that sheet flow is occurring with localized concentration. For the most part all slopes supported healthy vegetation, including many of the south southeast and south southwest facing slopes which are steeper than 3H:1V.

4.2.2 Landform Desktop Study

Catchment Areas

To complement the field observations, an investigation into the natural landforms in the general area of the Minto Mine was performed using GIS techniques. The investigation began with a review of the undisturbed catchment areas and the associated flow routing for each catchment area on site. It will be important to attempt to emulate these catchments in the reclamation designs for site landforms so as to ensure landforms are in equilibrium with landscape-wide flow patterns.

Groundwater flow in the Minto area is topographically driven and is generally towards Minto Creek, with recharge occurring at higher elevations and discharge occurring in valley bottoms. Reclaimed landforms at the Minto Mine should strive to maintain these flow directions in order to be hydrotechnically and erosionally stable. Catchment areas for the Minto Mine site are shown in Figure 5.

Slope Aspect

Slope directions, also known as slope aspects, were analyzed across the Minto Mine site and are pictured in Figure 6. In general, the mine lies in an area with east-west trending low mountain ranges that have significant variability in slope aspects across the site due to the curvilinear natural ridgelines that cause the slope aspects to vary along each slope.

The natural variability in slope aspect across the site is strongly correlated with vegetation type, which will be discussed further in sections below. The introduction of similar variability in slope aspect to reclaimed landforms at the Minto site wherever practical may lead to a wider variety of vegetation that is successfully established, and may ultimately lead to a more natural landscape over the long term post-closure.

Slope Gradient

From Figures 7 through 11, it can be seen that the majority of slopes in undisturbed areas are typically shallower than 18° (3H:1V), whereas slopes in mine-impacted areas are typically closer to 27° (2H:1V) for waste dumps or significantly steeper in the open pits. Figure 2 shows that areas close to natural ridgelines are often between 8° (7H:1V) and 14° (4H:1V), but away from the ridgelines the natural slopes become shallower than 6° (10H:1V). This is known as a concave slope profile, in which the slope is steeper at higher elevations and shallower at lower elevations, as shown in Figure 4. The analysis shows the significant variability in the natural slope angles.

Landform Patterns

Landform analysis found a correlation between elevation and surficial geology and a weaker (but still useful) correlation between slope aspect and vegetation type respectively.

The correlation between elevation and surficial geology is caused by the fact that certain soil types are stable only at certain slope angles, and as can be seen from Figure 2 and Figures 7 through 11, slope angles are also correlated with elevation in that slopes closer to crests and higher in elevation tend to be steeper than those farther from slope crests at lower elevations. At higher elevations and steeper slope angles, surficial geology tends to be either bedrock outcrop or a thin colluvial soil cover. At lower elevations and flatter slope angles, thicker sequences of glacial, periglacial, aeolian, and fluvial deposits are present, as pictured in Figure 12.

The correlation between slope aspect and dominant vegetation type is a function of the amount of sunlight experienced at different slope aspects, but is complicated by the recent fire history and subsequent vegetative succession. In general, southern aspects experience significantly more sunlight than northern aspects, which leads to decreased soil moisture contents, earlier snow melts, increased soil temperature and increased photosynthesis. These items combine to favour significantly different dominant species depending on the slope aspect, as can be seen in Figure 13. The present distribution of vegetation does not reflect the climax species mix at all locations due to the succession following various fires that have occurred at different times, and this likely is a factor in the weaker correlation between current vegetation and aspect.

5 Reclamation Land Units

The development of appropriate reclamation land units (RLUs) for the Minto Mine will facilitate achievement of the following outcomes:

- Geotechnical stability of reclaimed slopes;
- Minimal surface erosion by water and wind;
- Maintenance of acceptable long term water quality objectives, especially with respect to suspended solids; and
- Creation of conditions that will allow for re-colonization by native plant species, for basic ecological functions, and for use by local wildlife populations.

5.1 RLU Delineation

Delineation of reclamation land units is often based on end land use, initial and final surface configuration, and likely vegetation patterns (Bowman and Baker 1998). It is useful therefore to understand the land and vegetation patterns and existing biogeoclimatic zones defined within the given project area. Maps can be a valuable tool for achieving this. Biogeoclimatic zones are generally defined by superposition of the terrestrial ecozones and ecoregions, ecoclimatic regions, and the bioclimatic zones of a specific target area. The Minto Mine area is characterised as part of the Yukon Plateau ecoregion within the Boreal Cordillera ecozone (Figure 14), while from an ecoclimatic perspective, the mine area is part of the Northern Cordilleran High Boreal region (Figure 15).

Detailed biogeoclimatic zonation is not readily available for the Minto Mine area; however, extrapolation of biogeoclimatic zones onto the Minto site can be roughly performed based on the bioclimate zones determined for West Central Yukon (Figure 16) corroborated with the ecoclimatic description provided above.

Based on the above considerations, the vast majority of the Minto site falls within the Boreal High (BOH) biogeoclimatic zone, with some higher elevation areas possibly falling within the Subalpine (SUB) zone and some lower elevation areas falling within the Boreal Low (BOL) zone. In general, the BOH zone occurs at middle to upper elevations, is usually forested by black or white spruce, and is characterized by short, cool, and moist summers with long, cold winters. The SUB zone occurs at higher elevations on steep slopes and rocky highlands above the BOH and is sparsely forested or non-forested. The BOL zone occurs below the BOH zone at lower elevations along major river valleys and is usually forested by spruce and aspen with moderate understories. Vegetation is similar to the BOL zone but the warmer climate often results in much larger trees (McKenna *et al*. 2010).

5.1.1 End Land-use Designations

In the absence of published biogeoclimatic zones for the Minto site, reclamation land unit delineation was primarily based on End land-use designations. End land-use designations were defined by analyzing the native vegetation mapping data available for the Minto area as follows:

- Rocky Slope: Defined as slopes having little vegetation, coarse, thin soil veneers, and slopes steeper than 50% (27°, or 2H:1V). Roughly 6% of land in the Minto area falls into this category;
- Wetland: Defined as having flat slopes (<1%), thick vegetation, and standing water with organic soil. Less than 1% of land in the Minto area falls into this category;
- Forested: Defined as areas having primary vegetation types of Trembling Aspen, White Spruce, Black Spruce, Lodgepole Pine, or Alaskan Birch. Approximately 59% of land in the Minto area falls into this category; and
- Deciduous Shrubland: Defined as areas having primary vegetation types of Willow, Alder, or Scrub Birch. Approximately 35% of land in the Minto area falls into this category.

5.1.2 Slope Aspect

RLUs were then divided based on slope aspect category. Slope aspects can be categorized in a number of ways (France 2007; Gelbard and Harrison 2003); for the Minto site warm slopes were defined as having an aspect between 135^{o} and 270^{o}, cool slopes were defined as having aspects between 330^{o} and 90^{o}, and remaining aspects were defined as neutral. A schematic of this distribution is presented in Figure 17.

Flat slopes (defined as having a slope gradient less than 3°) were considered as a separate slope aspect category (Figure 18). Although in some cases aspect of these flat slopes may be directionally consistent with the cool slope, in practical terms these are warm slopes as direct solar radiation is received for most of the day.

5.1.3 Elevation, Slope Gradient, and Biogeoclimatic Zones

Four categories were defined based on the likely biogeoclimatic zones and the range in both slope gradient and elevation in the Minto area. These categories were formed based on the vegetation that the landscape in the Minto area will likely be able to support, and are as follows:

- BOL – Low elevations and flat slopes;
- BOH – Low to middle elevations and shallow to moderate slopes ranging from 1 to 25%;
- BOH – middle to high elevations and moderate to steep slopes ranging from 25-50%; and
- SUB – high elevations and steep slopes over 50%.

5.2 RLU Categories

Table 2 displays the RLUs for the Minto Mine site. It should be noted that these RLUs are conceptual only and have not yet been applied to specific areas of the site.

Table 2: Reclamation Land Units for the Minto Site

End Land-Use	Biogeoclimatic Zone	BOL	BOH	BOH	SUB
	Elevation	**Low**	**Low-Mid.**	**Mid.-High**	**High**
	Slope	**Flat (<1%)**	**Shallow-Mod. (1-25%)**	**Mod.-Steep (25-50%)**	**Steep (>50%)**
	Slope Aspect				
Forested	Cool and Neutral	-	FLm_{CN}	-	-
	Warm	-	FLm_{W}	FMh_{W}	-
Deciduous Shrubland	Cool and Neutral	-	SLm_{CN}	SMh_{CN}	-
	Warm	-	-	SMh_{W}	SH_{W}
Wetland	All	WL*	-	-	-
Rocky Slope	All	-	-	-	RH

* The first letter of each RLU denotes the end land-use category. The second and third (for some) letters denote the elevation category (i.e. L for low, Lm for low to middle). The subscript letters stand for slope aspect.

Figures 18 and 19 display the slope aspect and angle categories overlain by vegetation types that were used to define the RLUs for the Minto site. Figure 20 is a combination of slope aspect and angle in relation to vegetation. Mining-impacted areas where the slope differs significantly from the surrounding landscape, shown in black, will likely require significant resloping.

6 Discussion

6.1 Use of Site Specific Reclamation Land Units

In natural catchments, slope gradients develop over time as a function of prevailing substrate and hydrologic conditions. The type and amount of vegetation established on slopes depends heavily on the soil moisture availability of those slopes, which greatly depends on substrate, slope gradient, and slope aspect. Slopes on reclaimed landforms at the Minto Mine will likely experience less erosion and greater vegetation establishment if they are graded similarly to the surrounding natural slopes.

RLUs and landform design concepts have been defined for the Minto site based on site-specific features and the current mine plan. As progressive reclamation activities proceed according to the principles described in this document, it is expected that there will need to be opportunities for refinements to the design concepts, RLUs, and monitoring processes based on lessons learned during progressive reclamation and monitoring of performance indicators. As such, this should be considered a “live document”.

The RLUs defined in this document are conceptual and still need to be applied to specific landforms on the Minto site. It is important to work within the framework of the RLU categories from the beginning of progressive reclamation until final closure of specific landforms so that performance may be monitored accordingly and costs can be tracked to inform future closure planning. Progressive reclamation activities have already begun at the Southwest Waste Dump and RLUs are required there. The Main Waste Dump and the Dry Stack Tailings Facility will require RLU designations as soon as both facilities are complete or nearing completion, with initial progressive reclamation underway and final reclamation activities expected to begin in the near future.

The approach outlined in this report provides a basis for developing closure landform designs for each of the specific waste storage facilities at Minto Mine.

6.2 Reclamation Costing Considerations

While the RLUs were delineated based on various physical aspects of the site, they were also delineated in order to provide a framework for closure and reclamation costing. Specific land treatments can be developed for each RLU as part of the closure design detailing the amount of effort necessary to both reslope and revegetate mine-impacted landforms. Average costs may then be calculated to apply those treatments to each RLU, thus creating a unit cost basis for waste facilities and disturbed land reclamation.

For example, areas needing significant resloping will be more expensive to reclaim than those needing little or no resloping. As well, it is reasonable to assume that more effort will be necessary to revegetate cool and neutral forested slopes than warm slopes covered with shrub, so separate RLUs were defined accordingly. As progressive reclamation takes place within a specific RLU, actual completion costs can be recorded and projected for future reclamation activities.

6.3 Landform Monitoring

Monitoring should occur within the framework of the defined RLUs such that landform performance is measured for each RLU. Monitoring should begin while progressive reclamation takes place so that RLU assignments may be adjusted based on performance as reclamation proceeds.

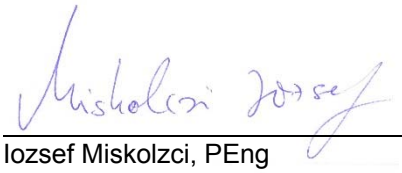
Monitoring should include the following actions:

- Erosion monitoring – set up erosion monitoring stations on each RLU after reclamation activities have taken place to estimate the annual erosion from constructed closure landforms and compare to both degree of revegetation and to erosion estimated during design for performance indicators;
- Water quality monitoring – monitor water quality from closure landforms for each RLU, specifically for suspended solids along with any other pertinent quality indicators, depending on the landform; and
- Vegetation surveys – periodic surveys of vegetation establishment on each RLU to ensure reclamation goals with respect to vegetation type and density are being met.

This report, Closure Landform Design and Reclamation Landform Units for the Minto Mine, was prepared by



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All data used as source material plus the text, tables, figures, and attachments of this document have been reviewed and prepared in accordance with generally accepted professional engineering and environmental practices.

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The opinions expressed in this report have been based on the information available to SRK at the time of preparation. SRK has exercised all due care in reviewing information supplied by others for use on this project. Whilst SRK has compared key supplied data with expected values, the accuracy of the results and conclusions from the review are entirely reliant on the accuracy and completeness of the supplied data. SRK does not accept responsibility for any errors or omissions in the supplied information, except to the extent that SRK was hired to verify the data.

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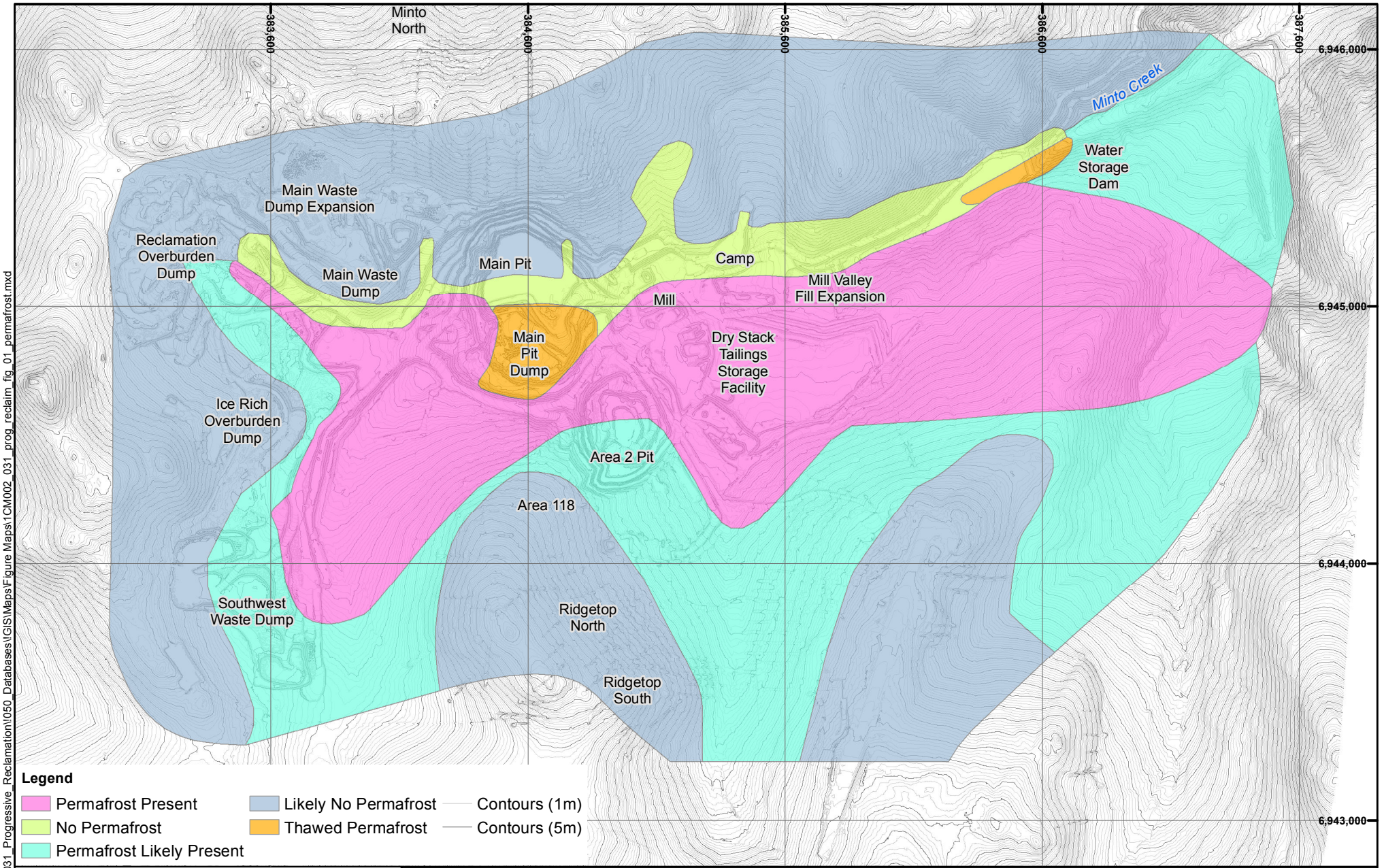
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Figures



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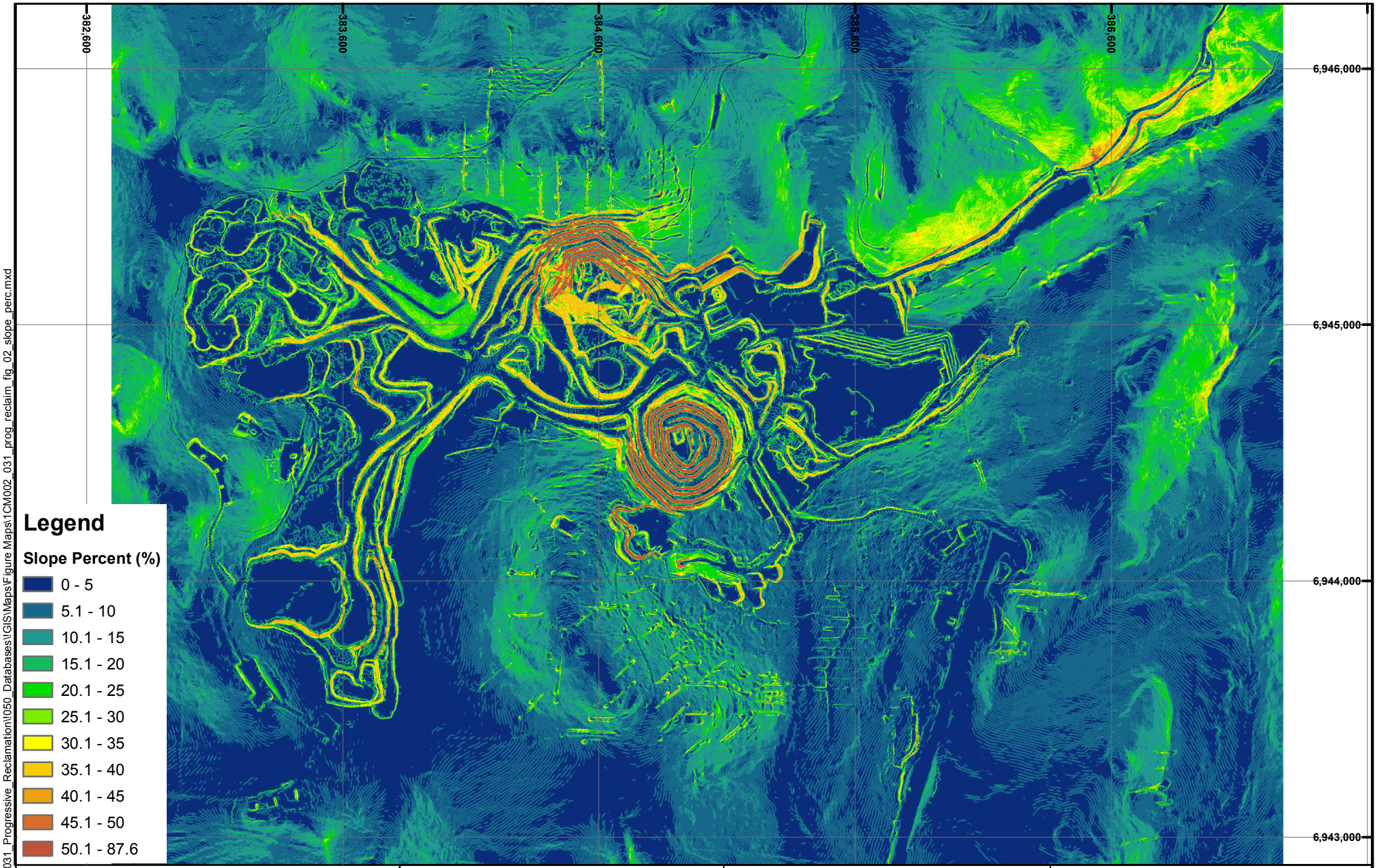
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Minto Mine

Closure Landform Design and RLUs

Permafrost Extent Across Site

Date:	Approved:	Figure:
April 2016	EKH	1



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Closure Landform Design and RLUs

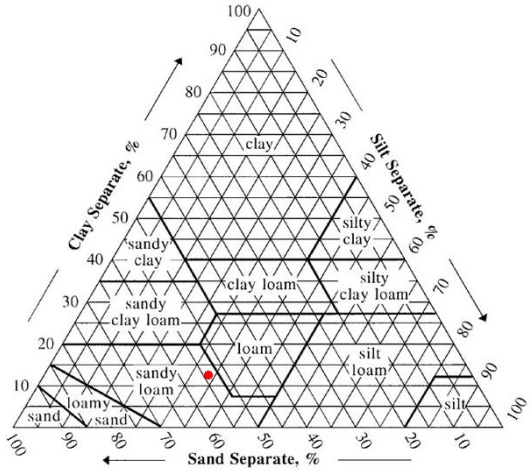
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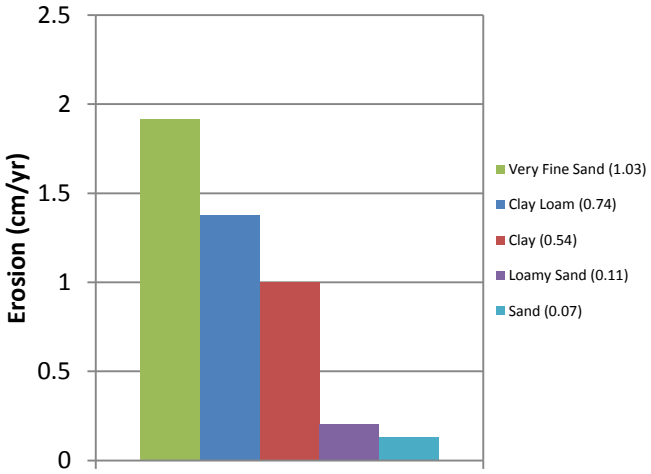
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Average particle size distribution of soil in the ROD falls into the Sandy Loam category (USDA classification)

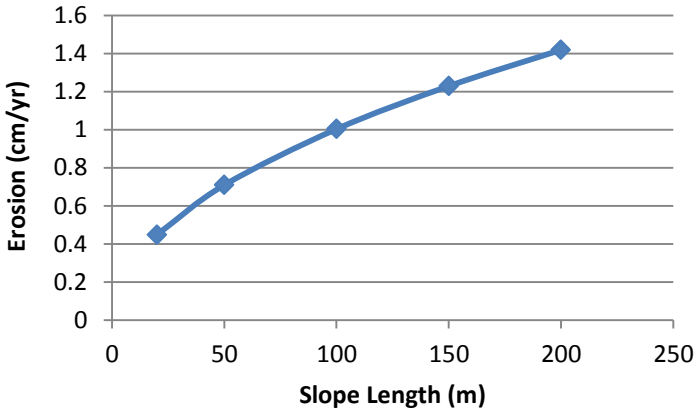
(a)

Erosion by soil type



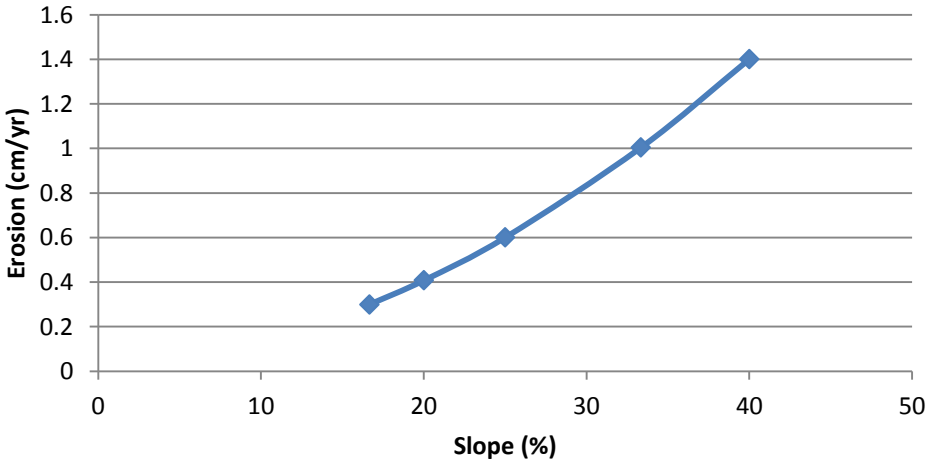
(b)

Slope length (m) vs Erosion (3H:1V)



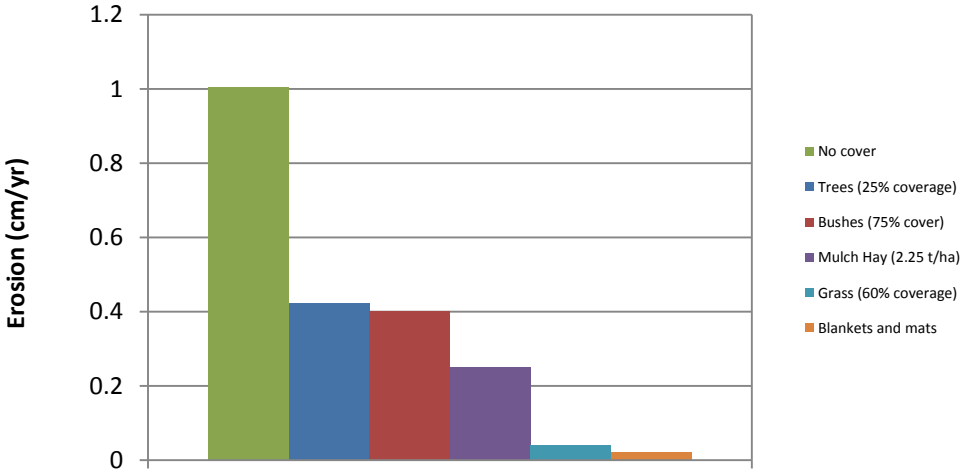
(c)

Slope (%) vs Erosion



(d)

Erosion by cover type



(e)

Note: Erosion modeling using USLE assumes uniform soil loss arising from sheet erosion – it does not address local gully erosion, and the results presented here are comparative rather than absolute. Density of soil was assumed to be 1.9 t/m^3 in calculations.

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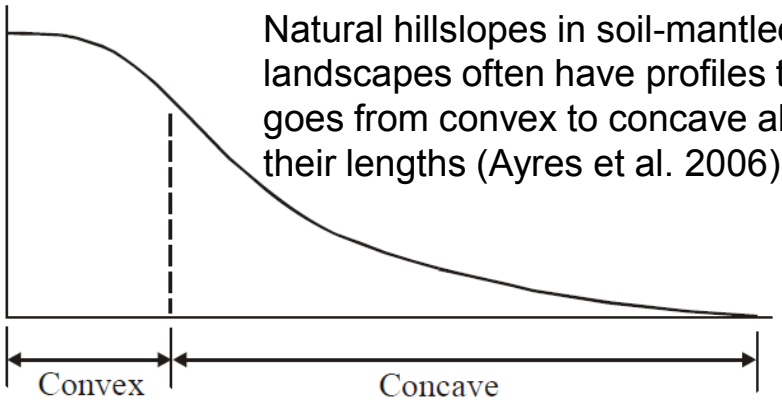
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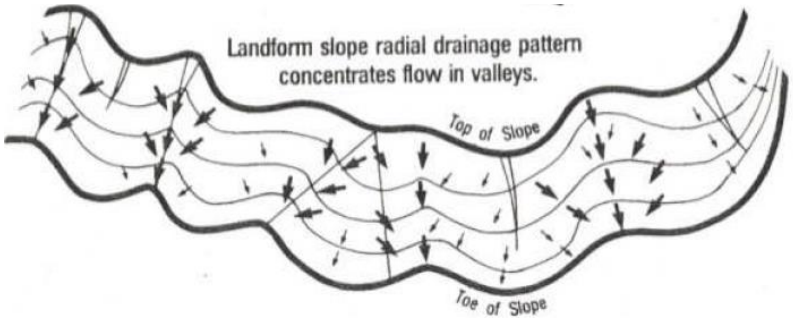
Closure Landform Design and RLUs

Erosion Analysis

Date:	Approved:	Figure:
April 2016	EKH	3



Natural hillslopes in soil-mantled landscapes often have profiles that goes from convex to concave along their lengths (Ayres et al. 2006).



Landform-designed slopes have a radial drainage pattern that concentrate flow in valleys (Michael et al. 2010).



Landform-designed slopes are broken up into smaller drainage areas in order to limit the maximum overland flow path and resulting erosion (Schor and Gray 2013).

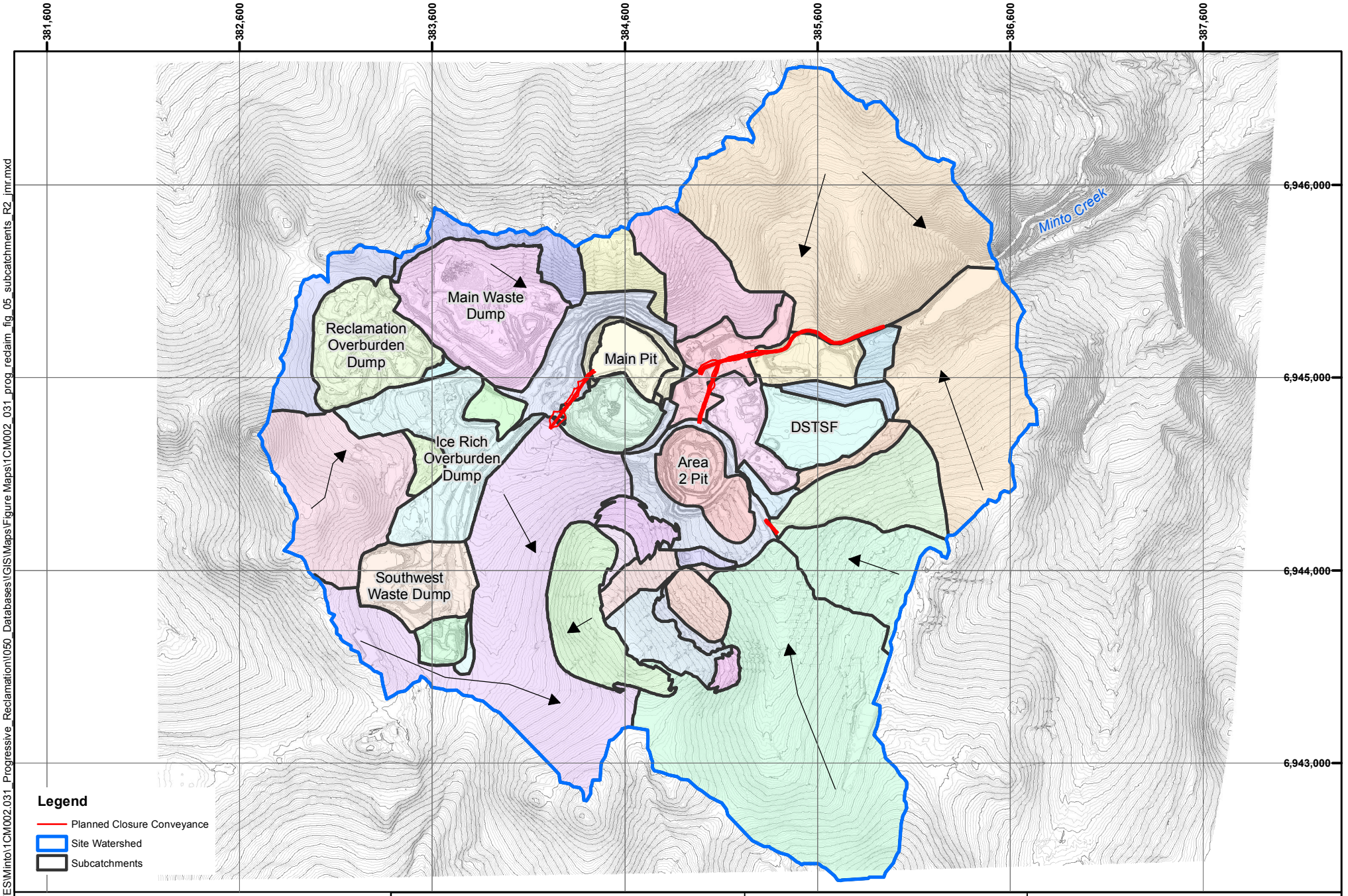
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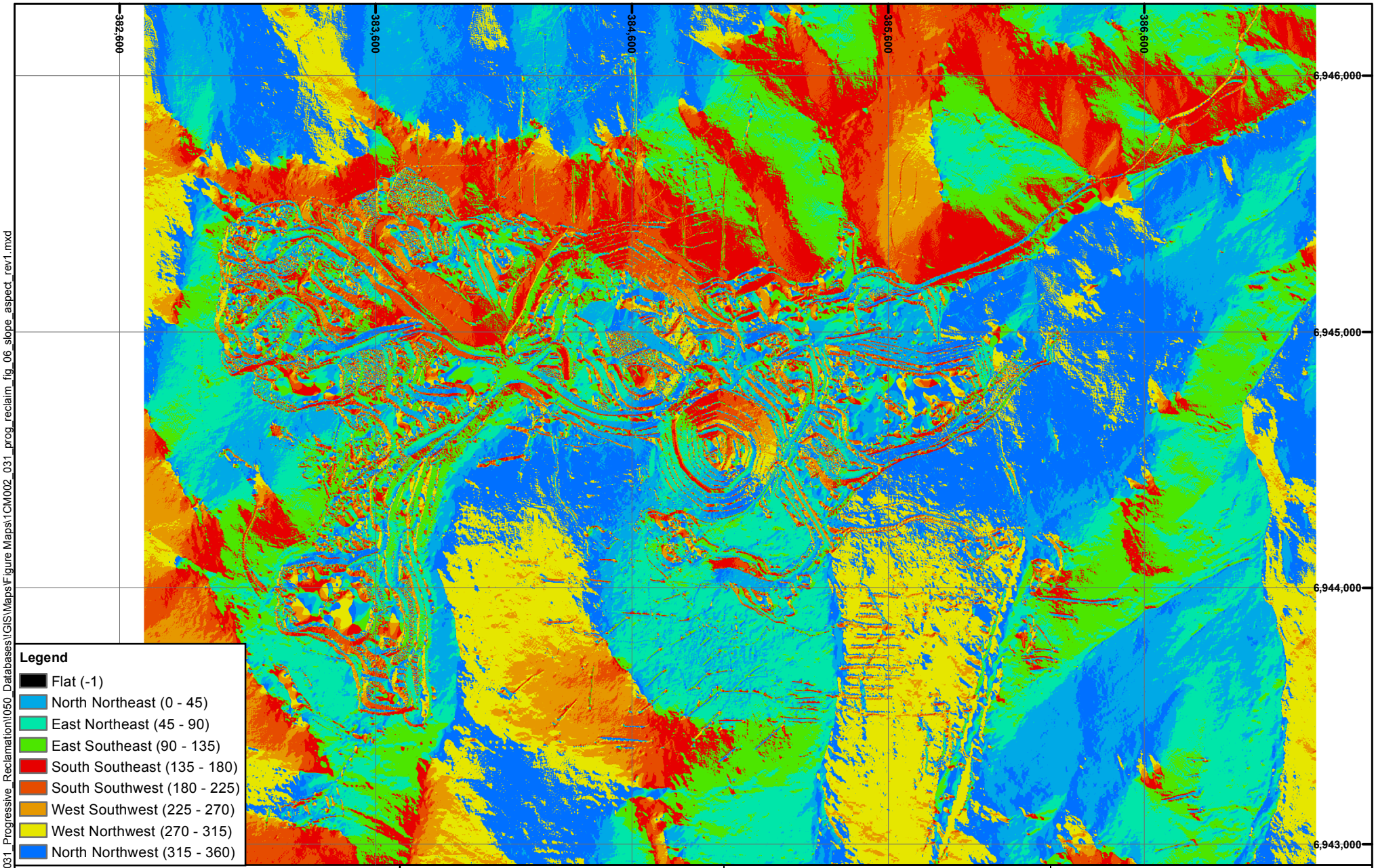
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Closure Landform Design and RLUs

Sub-Catchment Areas

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Closure Landform Design and RLUs

Slope Aspect Distribution

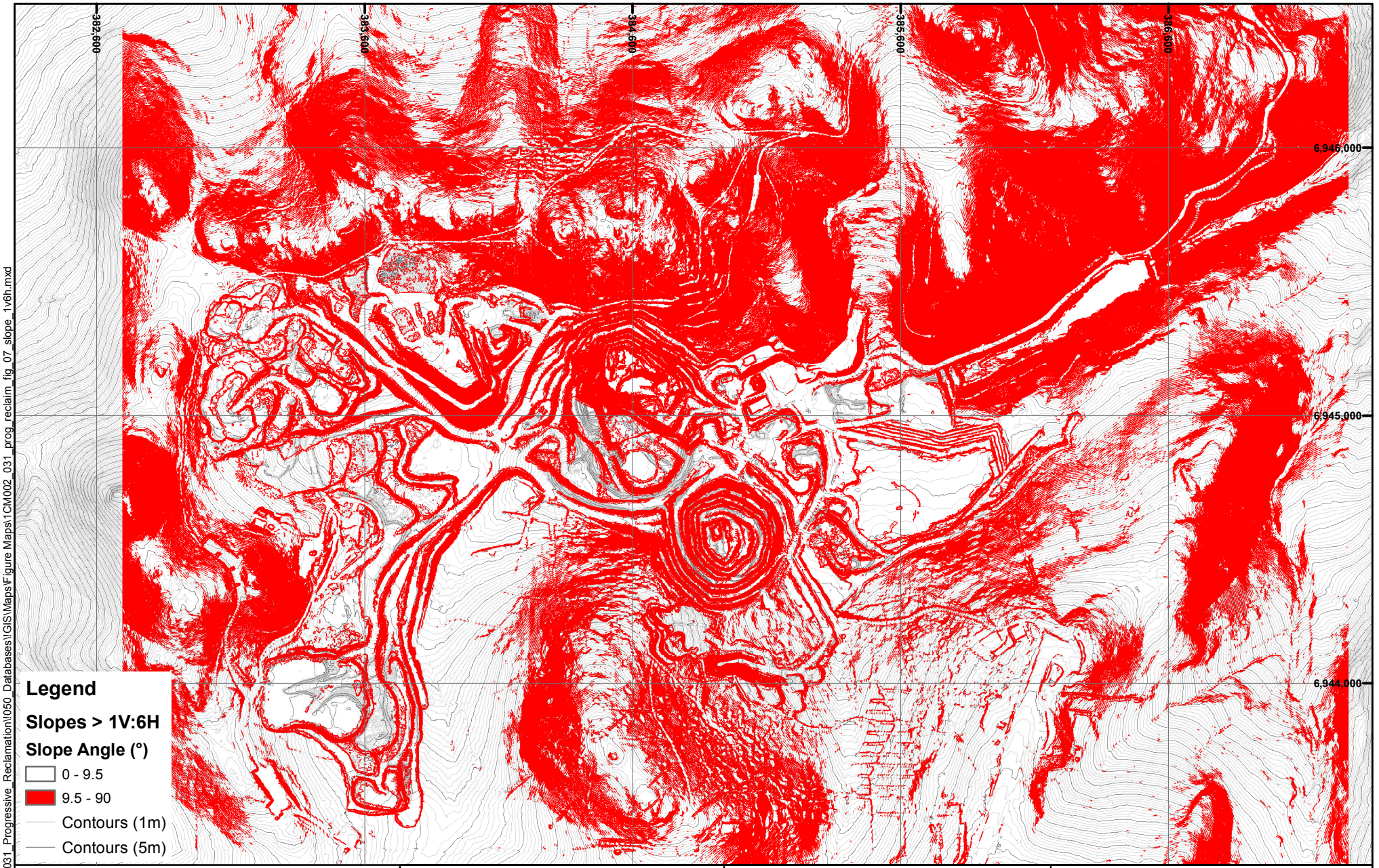
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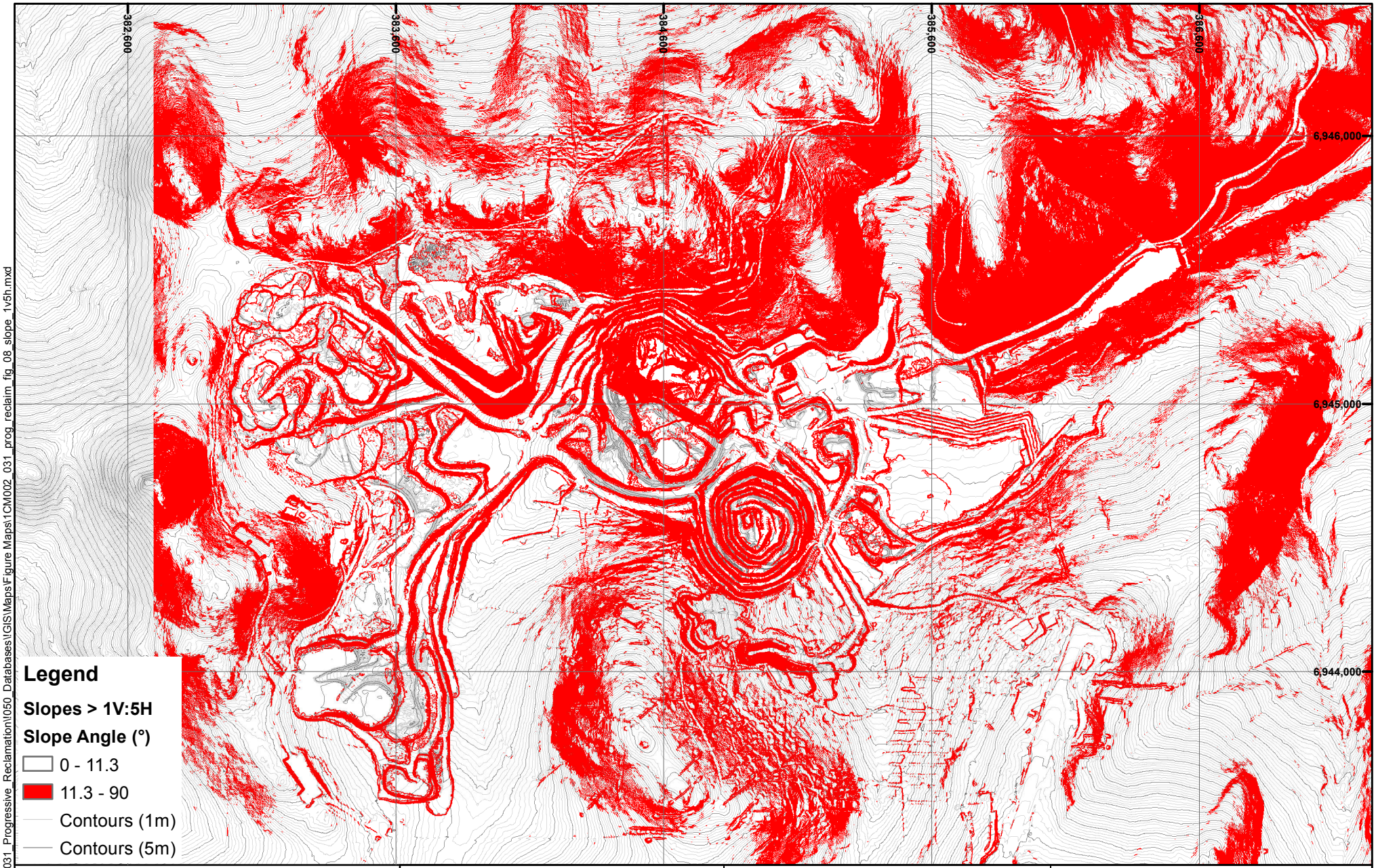
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Closure Landform Design and RLUs

Distribution of Slopes Steeper than 6H:1V

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Closure Landform Design and RLUs

Distribution of Slopes Steeper than 5H:1V

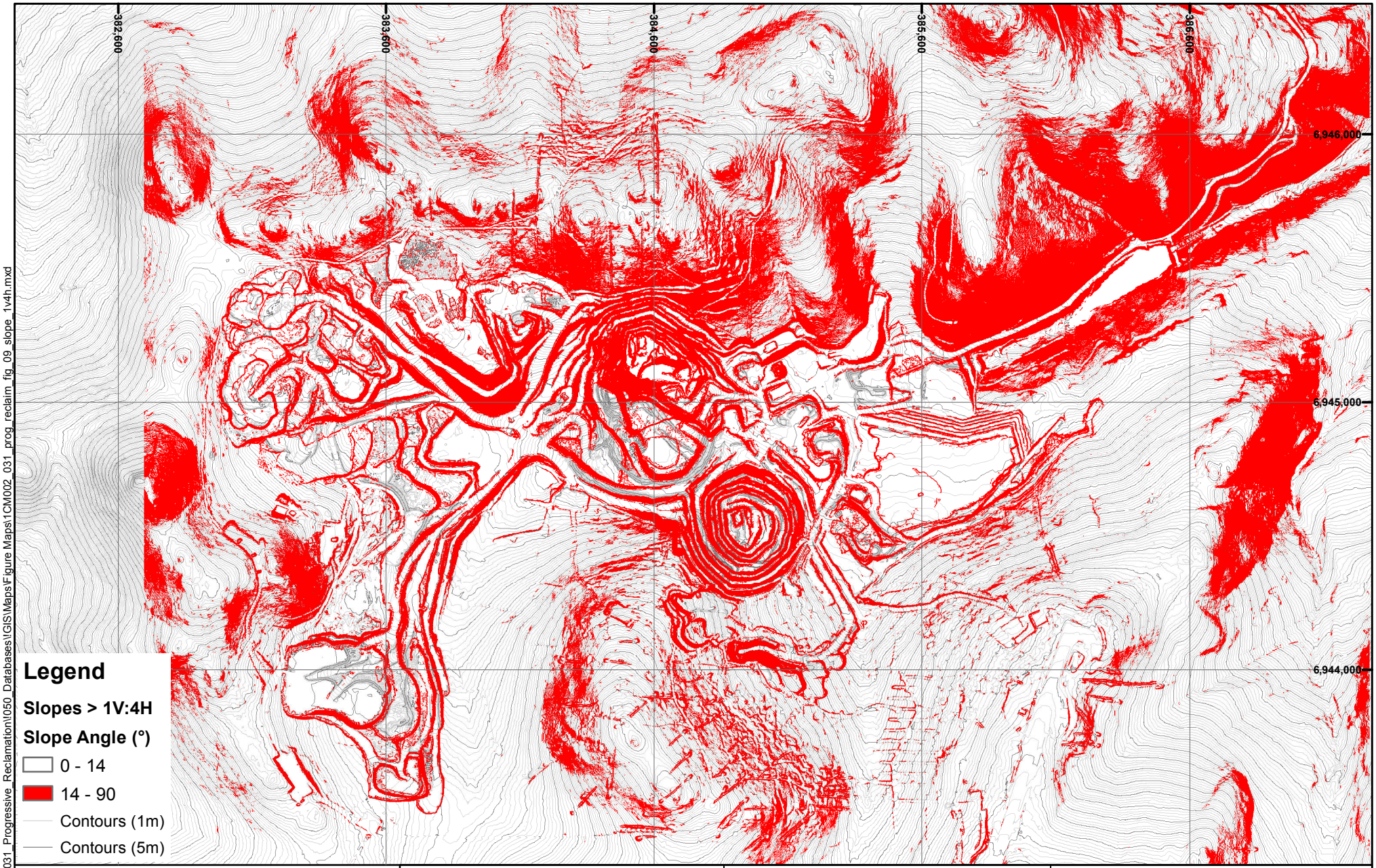
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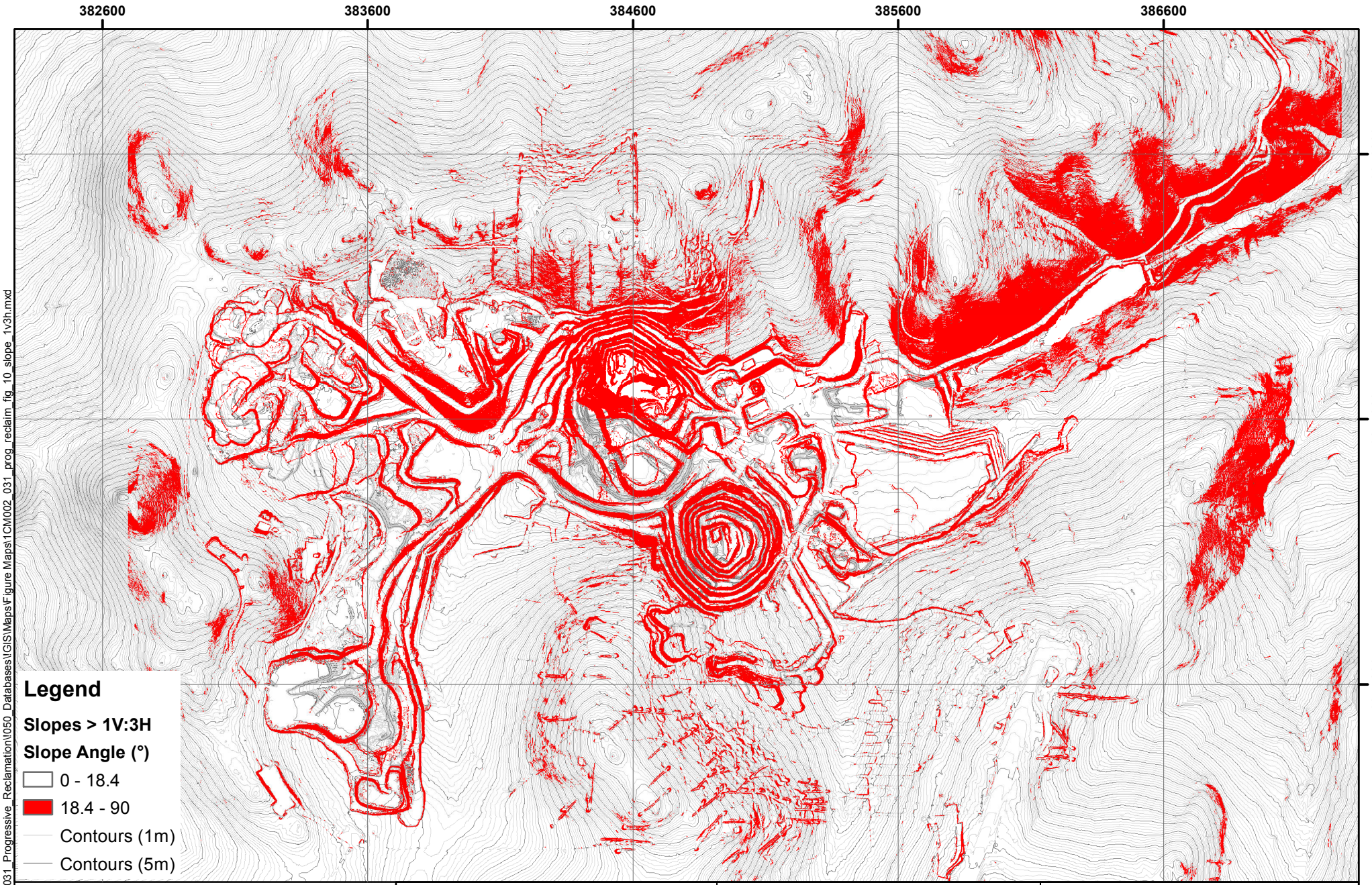
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Closure Landform Design and RLUs

Distribution of Slopes Steeper than 4H:1V

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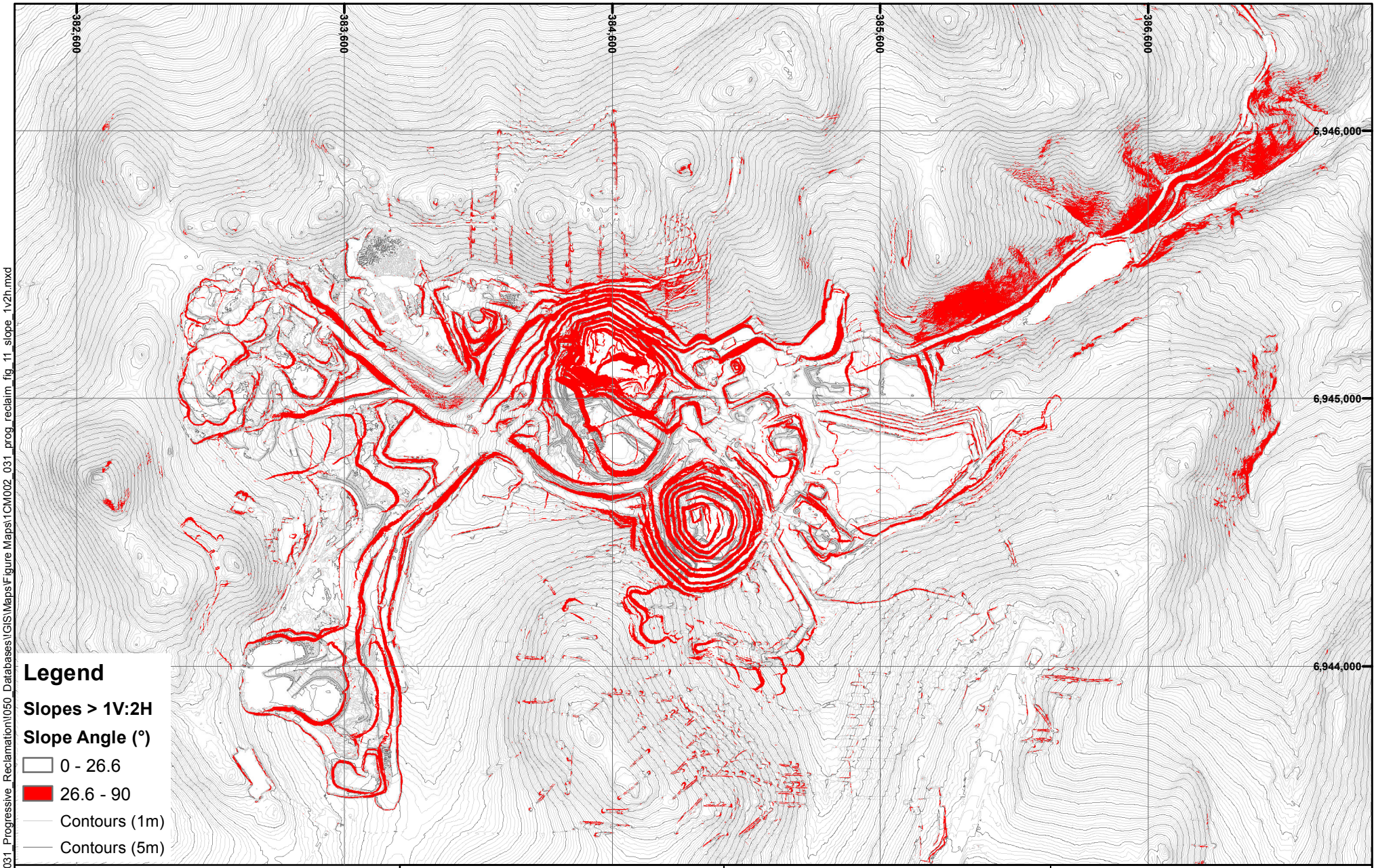
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Closure Landform Design and RLUs

Distribution of Slopes Steeper than 3H:1V

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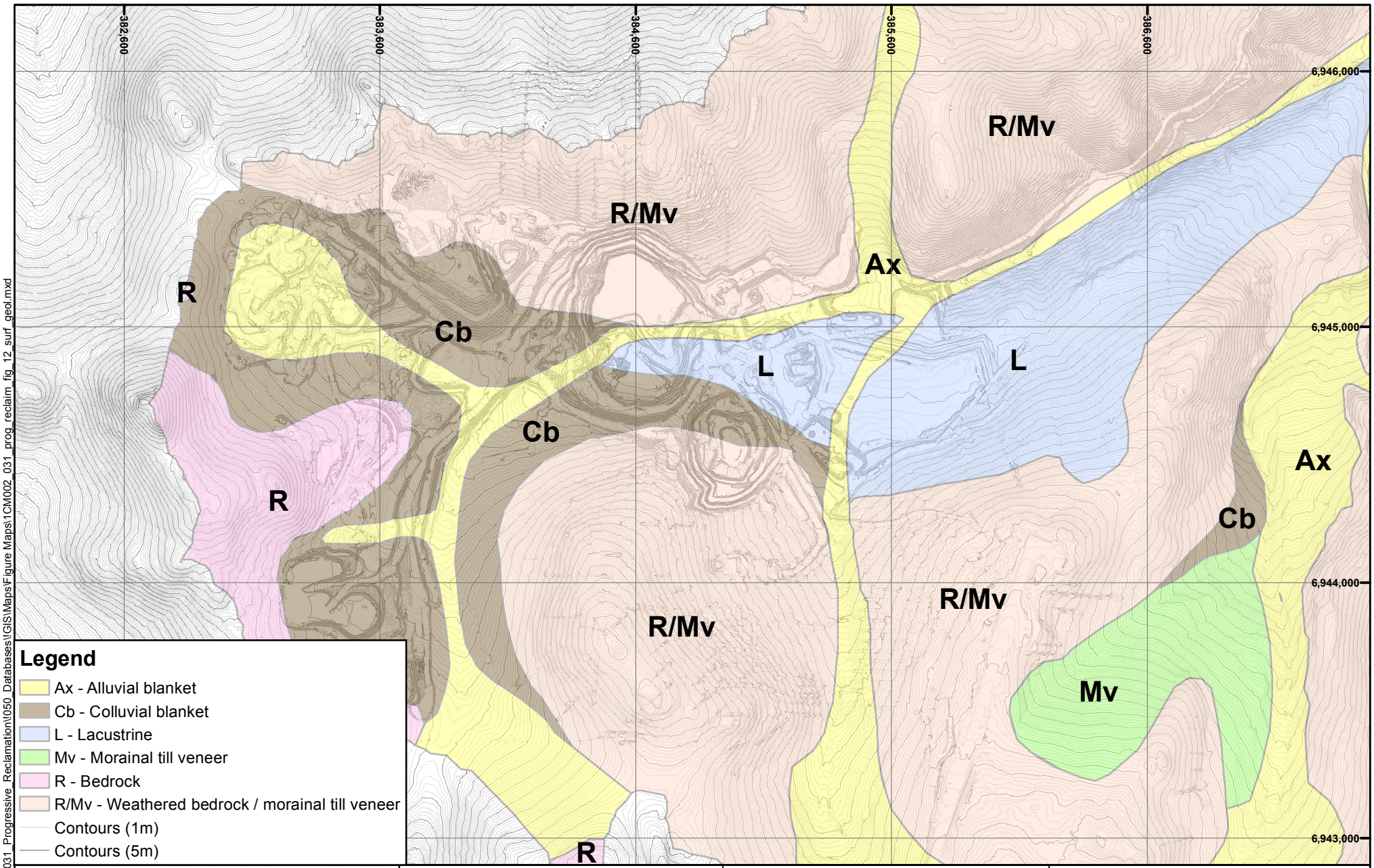
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Minto Mine

Closure Landform Design and RLUs

Distribution of Slopes Steeper than 2H:1V

Date:	Approved:	Figure:
April 2016	EKH	**11**



Source: SRK Consulting (US) Inc. (2014). Surficial Geology Map - Minto Creek Watershed

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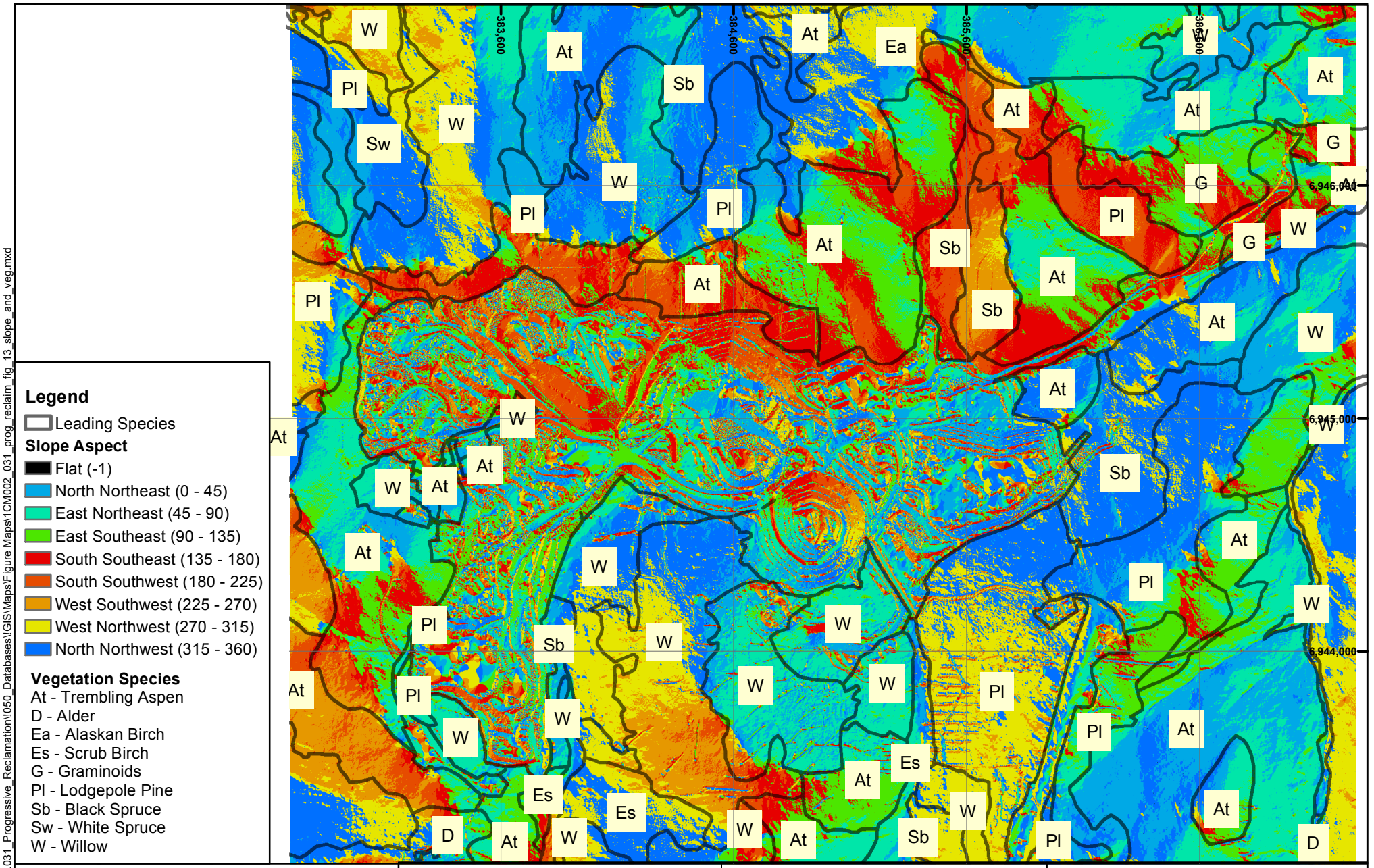
Minto Mine

Closure Landform Design and RLUs

Surficial Geology Map

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April 2016	EKH	12

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Minto Mine

Closure Landform Design and RLUs

Slope Aspect and Vegetation Categories

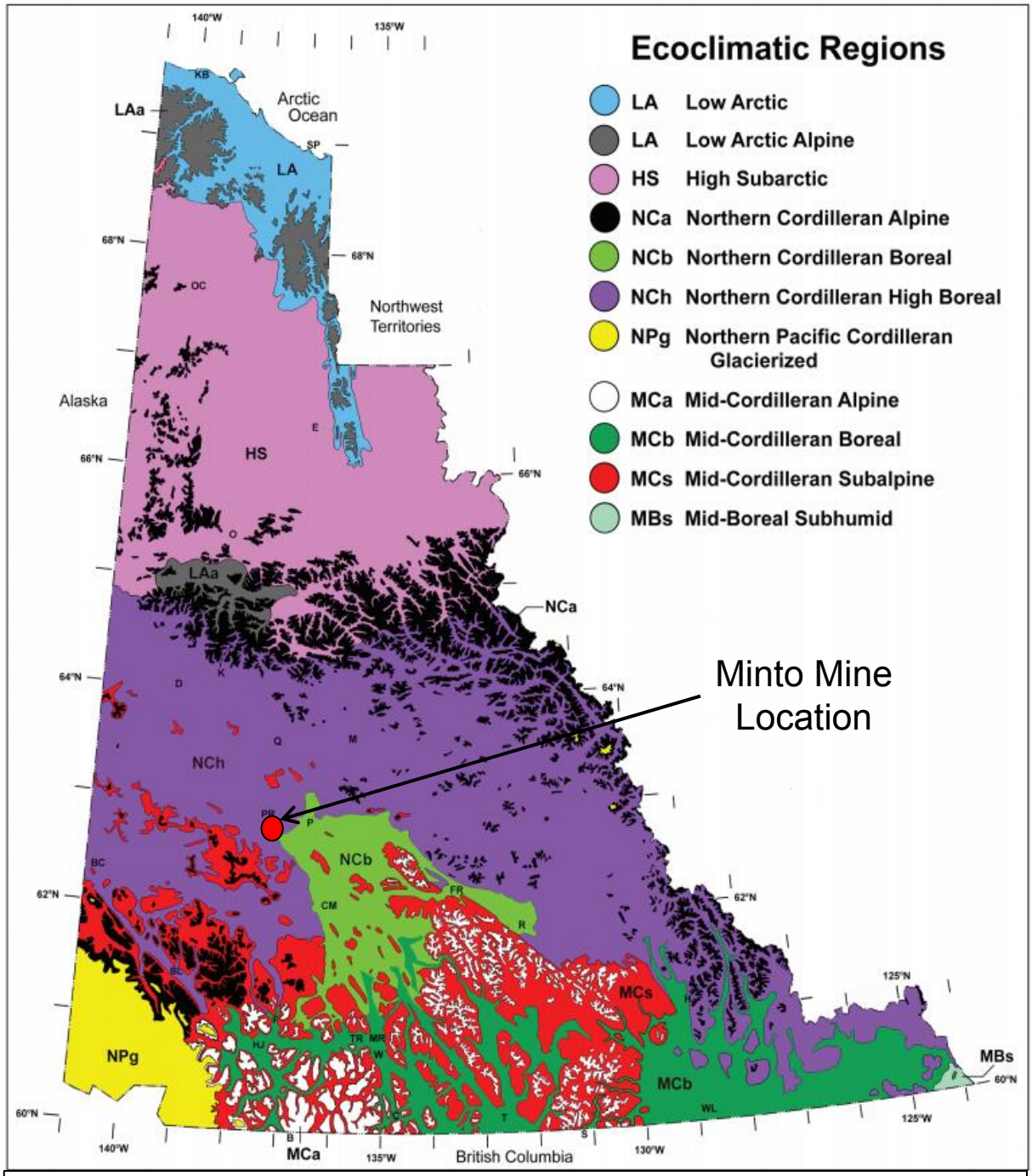
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Source: Smith, C.A.S., Meikle, J.C., Roots, C.F. (2004). *Ecoregions of the Yukon Territory: Biophysical Properties of Yukon Landscapes*. Agriculture and Agri-Food Canada, PARC Technical Bulletin No. 04-01, Summerland, British Columbia, 313 p. Retrieved from: http://www.env.gov.yk.ca/animals-habitat/documents/ecoregions_of_yukon_reduced.pdf

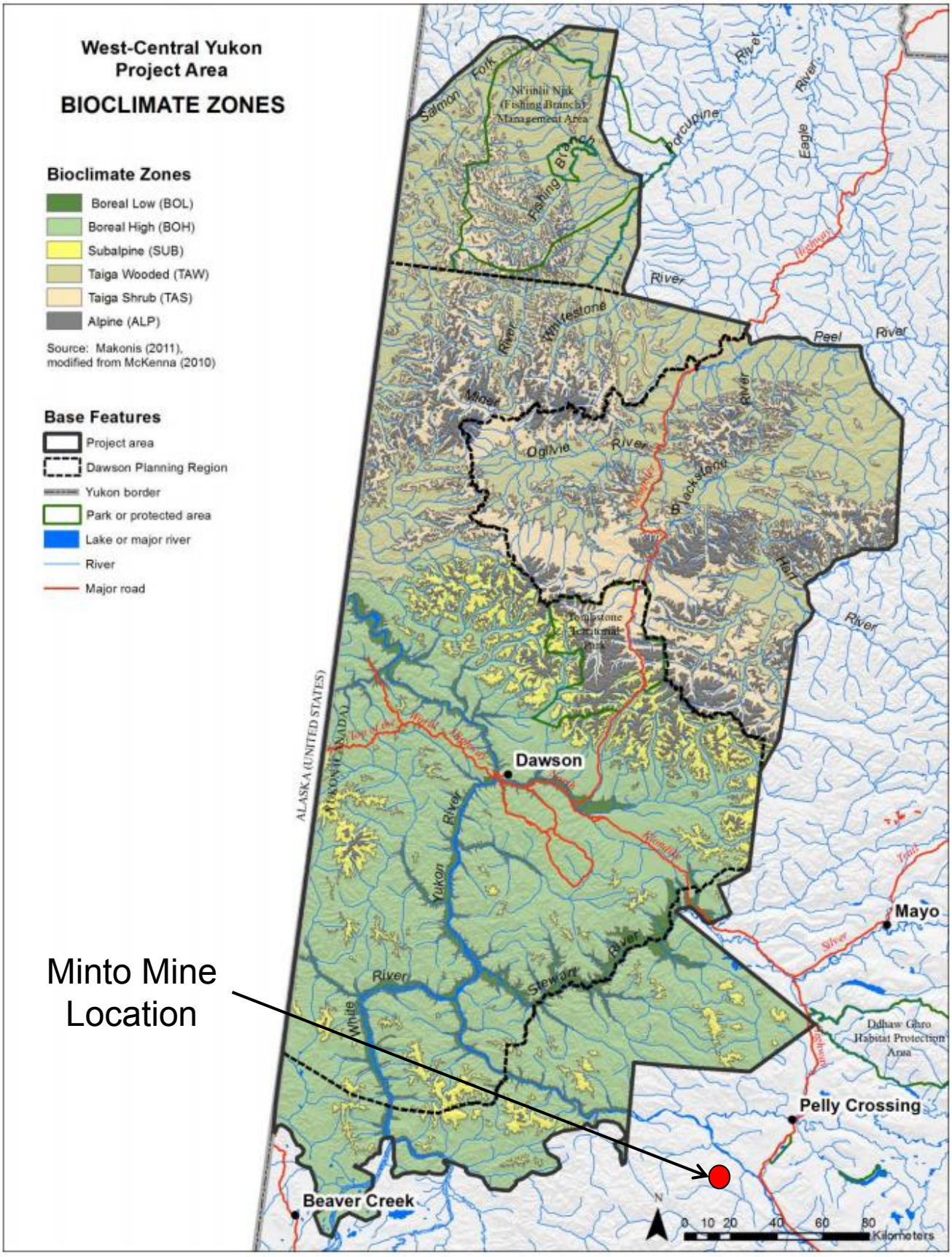
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		Terrestrial Ecozones of the Yukon
Job No: 1CM002.031 Filename: Fig14_TerrestrialEcozones_rev02.pptx	Minto Mine	Date: April 2016 | Approved: EKH | Figure: **14**



Letters other than ecoclimatic regions indicate the location of selected settlements and meteorological station locations: B – Blanchard River; BC – Beaver Creek; BL – Burwash Landing; C – Carcross; CM – Carmacks; D – Dawson City; E – Eagle Plains; F – Otter Falls; FR – Faro; H–Hour Lake; HJ – Haines Junction; K – Klondike; KB – Komakuk Beach; M – Mayo; MR – Mayo Road; O – Ogilvie River; OC – Old Crow; P – Pelly Crossing; PR – Pelly Ranch; Q – McQueston; R – Ross River; S – Swift River; SP – Shingle Point; T – Teslin; TR – Takhini Ranch; W – Whitehorse; and WL – Watson Lake.

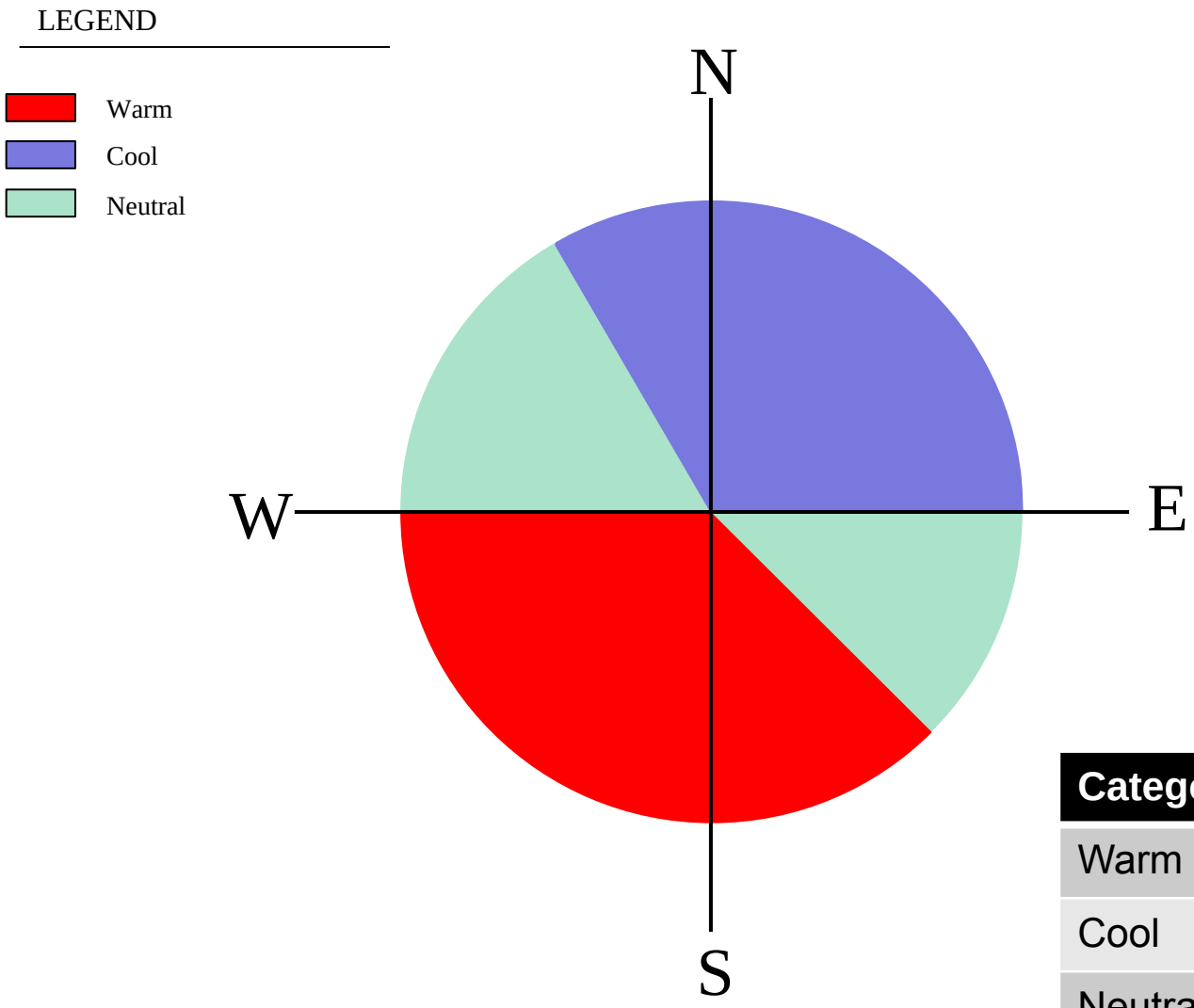
Source: W.L. Strong (2013) Ecoclimatic Zonation of Yukon (Canada) and Ecoclinal Variation in Vegetation

srk consulting	capstone MINTO MINE OPERATED BY MINTO EXPLORATIONS LTD.	Closure Landform Design and RLUs
		Ecoclimatic Regions of the Yukon
Job No: 1CM002.031 Filename: Fig15_EcoclimaticZones_rev02.pptx	Minto Mine	Date: April 2016 | Approved: EKH | Figure: **15**



Source: Makonis Consulting Ltd. (2012) Regional Ecosystems of West-Central Yukon

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		Biogeoclimatic Zones of West-Central Yukon		
Job No: 1CM002.031 Filename: Fig16_BiogeoclimaticZones_rev02.pptx	Minto Mine	Date: April 2016	Approved: EKH	Figure: **16**



Category	Azimuth Range (º)
Warm	135 – 270
Cool	330 – 90
Neutral	90 – 135, 270 – 330

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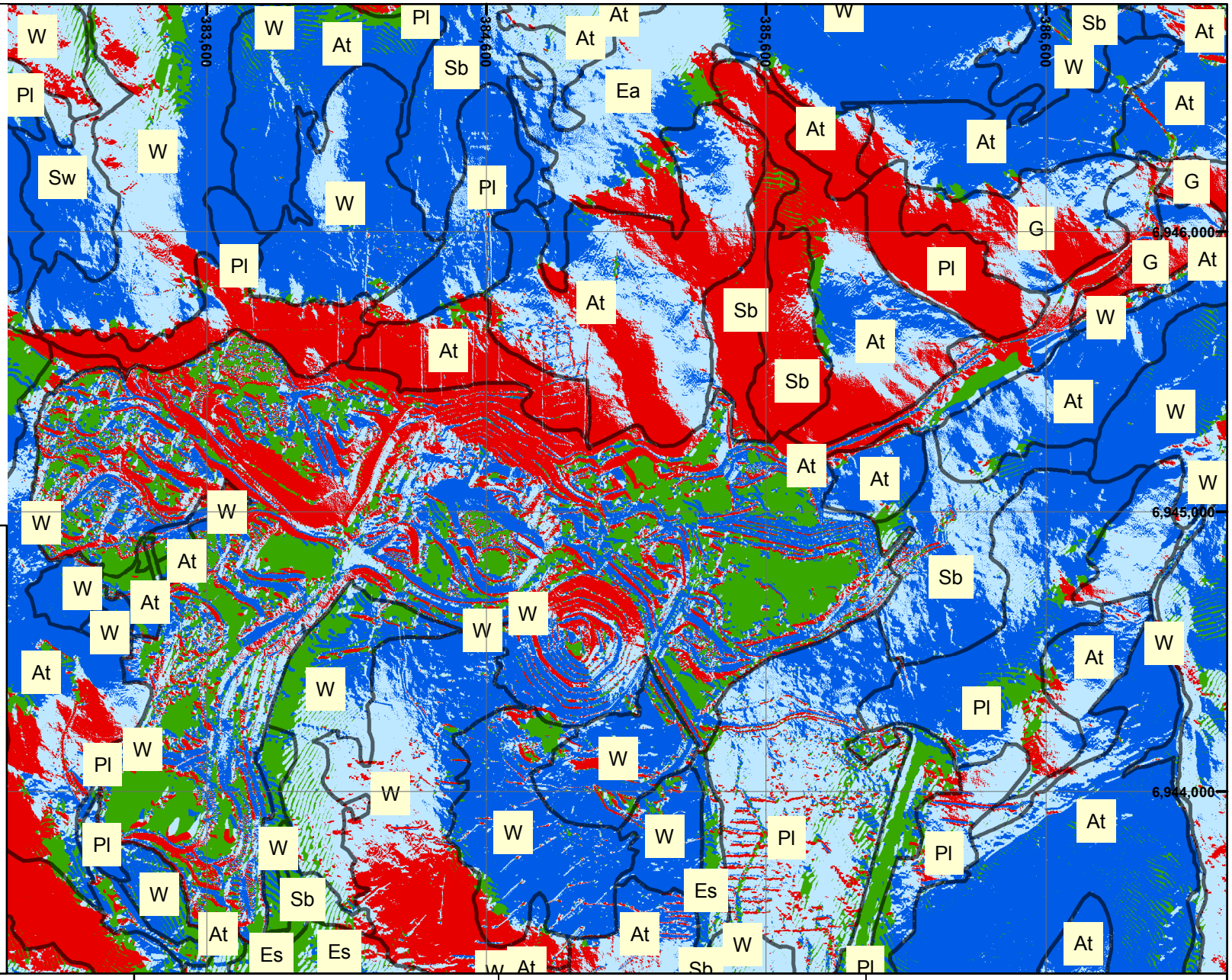
Job No: 1CM002.031
Filename: Fig17_SlopeAspectSchematic.pptx

Minto Mine

Closure Landform Design and RLUs

Slope Aspect Category Limits Schematic

Date:	Approved:	Figure:
April 2016	IM	17



Legend

Primary Species-Vegetation

Aspect

- Flat - All Aspects
- Sloped - Warm
- Sloped - Cool
- Sloped - Neutral

Vegetation Species

At - Trembling Aspen
D - Alder
Ea - Alaskan Birch
Es - Scrub Birch
G - Graminoids
Pl - Lodgepole Pine
Sb - Black Spruce
Sw - White Spruce
W - Willow

0 500 1,000 Meters

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Job No: 1CM002.031
Filename: 1CM002_031_prog_reclaim_fig_18a_slope_angle_veg

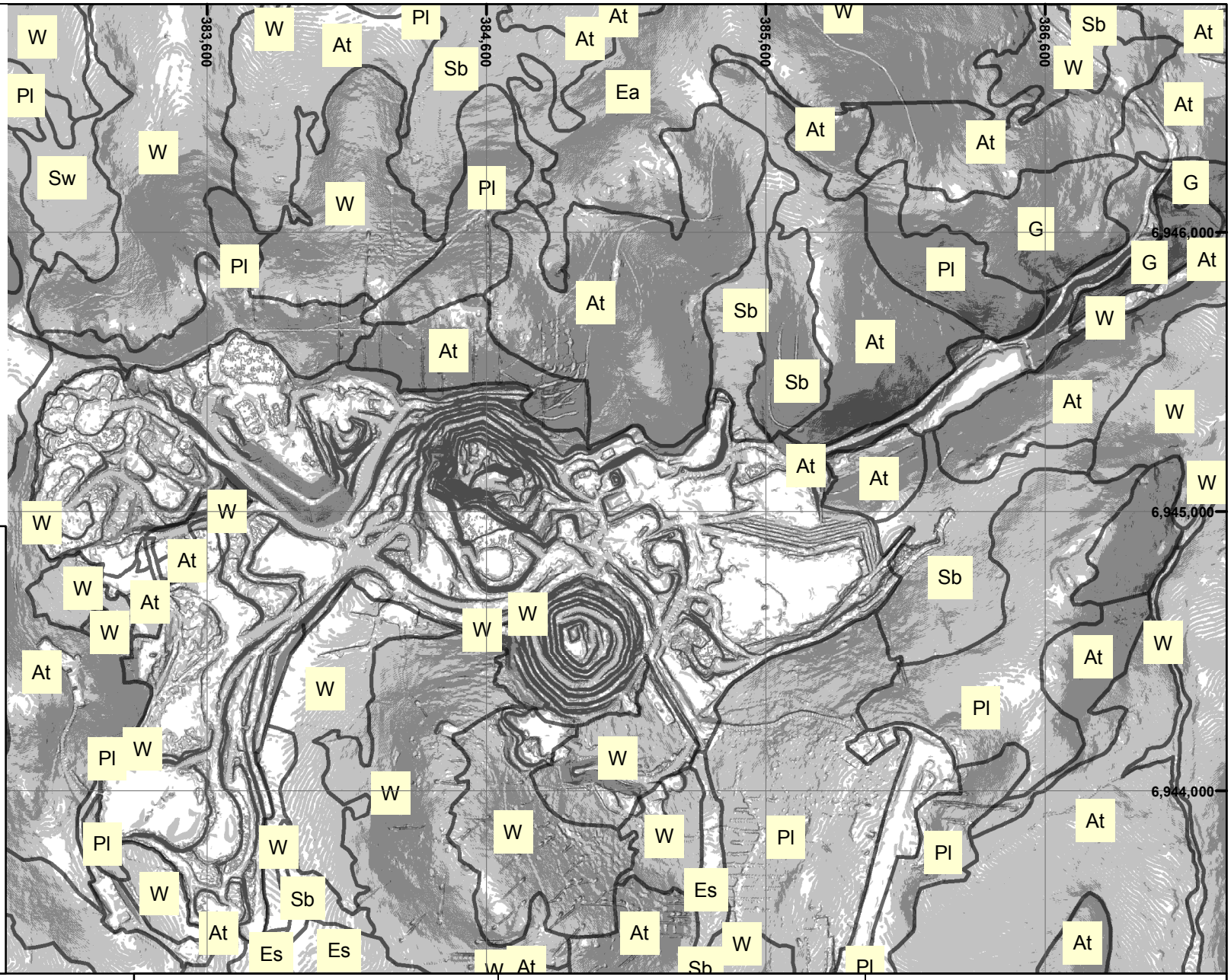
Minto Mine

Closure Landform Design and RLUs

Vegetation Categories and Slope Aspect

Date:	Approved:	Figure:
April 2016	EKH	18

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Legend

Primary Species-Vegetation

Slope (degree)

- Flat (0° to 3°)
- 6H : 1V (3° to 10°)
- 4H : 1V (10° to 14°)
- 2H : 1V (14° to 27°)

Vegetation Species

At - Trembling Aspen
D - Alder
Ea - Alaskan Birch
Es - Scrub Birch
G - Graminoids
Pl - Lodgepole Pine
Sb - Black Spruce
Sw - White Spruce
W - Willow

0 500 1,000 Meters

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Job No: 1CM002.031
Filename: 1CM002_031_prog_reclaim_fig_18b_slope_angle_veg

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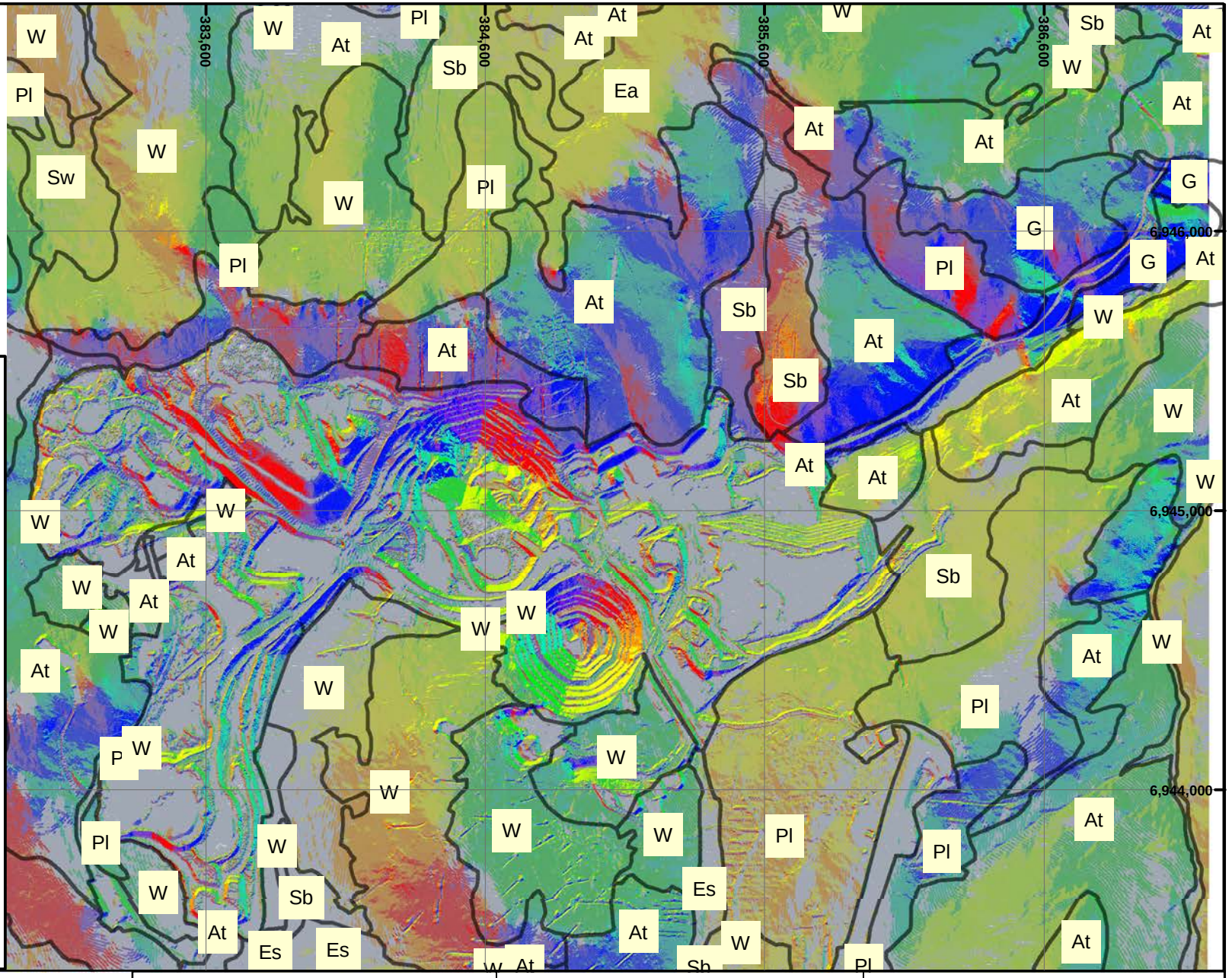
Minto Mine

Closure Landform Design and RLUs

Vegetation Categories and Slope Angle Categories

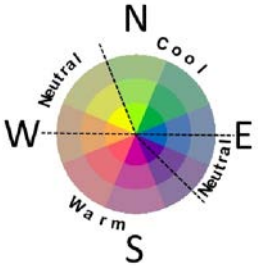
Date:	Approved:	Figure:
April 2016	EKH	19

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Legend

Aspect / Slope



Color Map
Inner Ring - Steep Slopes
Middle Ring - Moderate Slopes
Outer Ring - Shallow Slopes

Vegetation Species
At - Trembling Aspen
D - Alder
Ea - Alaskan Birch
Es - Scrub Birch
G - Graminoids
Pl - Lodgepole Pine
Sb - Black Spruce
Sw - White Spruce
W - Willow

0 500 1,000 Meters

srk consulting

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Job No: 1CM002.031
Filename: 1CM002_031_prog_reclaim_fig_19_slope_angle_veg

Minto Mine

Site-Wide RLU Designation

Slope Aspect, Angle, and Vegetation Categories

Date:	Approved:	Figure:
Nov 2015	JBK	**20**

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Background Water Quality of Lower Minto Creek for Application in the Derivation of Post-Closure Water Quality Objectives

Report Prepared For:
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July 2016

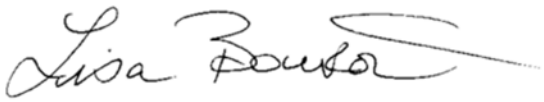
Background Water Quality of Lower Minto Creek for Application in the Derivation of Post-Closure Water Quality Objectives

Prepared for:

Minto Explorations Ltd. - Minto Mine

Prepared by:

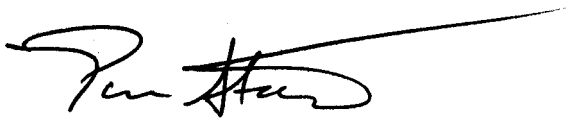
Minnow Environmental Inc.



Lisa Bowron, M.Sc.



Tim Barrett, Ph.D.



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APPENDIX D: WATER QUALITY DATA SUMMARY CALCULATIONS

1.0 INTRODUCTION

1.1 Site Description

The Minto Mine is a high-grade copper mine located within Selkirk First Nation (SFN) Category A Settlement Land Parcel R-6A approximately 240 km northwest of Whitehorse, Yukon Territory (62°37'N latitude and 137°15'W longitude; Figure 1.1). It is owned and operated by Minto Explorations Ltd., a wholly owned subsidiary of Capstone Mining Corporation. Development of the mine was initiated in 1997, commercial operations started in October 2007 and the currently anticipated operating life is to 2018. The facility is permitted to conduct open pit and underground mining, and includes a mill with a capacity of 4,200 tonnes of ore per day. The Minto Mine produced 16,500 kilo-tonnes of copper in 2015 (average copper grade 1.38%). Mineral reserves as of January 1st 2016 were approximately 5,500 kilo-tonnes at an average copper grade of 1.87%.

In addition to open pit mines, underground mines and the mill, the Minto Mine site includes a number of waste rock dumps, a concentrate storage shed, tailings storage facilities, a water retention dam with a water storage pond, a water treatment plant, administrative offices, an airstrip and a camp (Figure 1.2). Mill tailings are stored in dry stack tailings storage facilities and in mined-out open pits. Mine-impacted seepage from the tailings storage facilities and under the Mill Valley Fill Extension (MVFE) is collected at the Mill Valley Fill Extension Stage 2 Collection Sump and is pumped to the main pit (Figure 1.2). Non-impacted water and treated mine-impacted water are collected in a Water Storage Pond (WSP; Figure 1.2). Effluent from the WSP is periodically discharged to Minto Creek under conditions specified in the Minto Mine's Water Use Licence (WUL) QZ14-031 (August 2015; YWB 2015). Minto Creek, in turn, discharges to the Yukon River approximately 7.7 km south-east of the WSP (Figure 1.2).

The Minto Mine is actively planning for closure, which could potentially occur as soon as late 2018. Conceptually, closure will include three stages: 1) an active closure period; 2) post-closure 1; and 3) post-closure 2. All major reclamation work will be completed during the active closure period. This will include re-contouring, cover placement, re-vegetation, construction of closure water conveyance channels, construction of wetland water treatment systems, and building demolition. Water will be actively managed during the active closure period. Post-closure 1 follows the completion of all major reclamation work and represents the period when the closure measures completed during the active closure period become fully established (e.g., vegetation is established, cover material self-armors, constructed wetlands are established, water flow through closure conveyances is established). Water treatment during post-closure 1 will be both active and passive. Post-closure 2 follows post-closure 1 and final



N
W E
S

160 0 160

Kilometers

MAP INFORMATION
Datum: NAD 83 Map Projection: UTM Zone 8V
Data Source: Department of Natural Resources Canada
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Creation Date: June 2016
Project No. 167202.0062

Map of Canada



Figure 1.1: Location of the Minto Mine

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Features

- Final Effluent Discharge Point
- Waterbody
- Minto Creek Detention Structure
- Building
- Watercourse
- Mine Road
- Water Flow Direction

N
W E
S

0 175 350 700
Meters

MAP INFORMATION
Datum: NAD 83 Map Projection: UTM Zone 8V
Data Source: National Topographic Data Base (NTDB) compiled by Department of Natural Resources Canada at a scale of 1:50,000. All rights reserved.

Mine infrastructure data provided by Access Mining Consultants Inc.
Mine site contours derived from 2012 aerial imagery obtained from Challenger Geomatics.

Creation Date: June 2016
Project No. 167202.0062

Figure 1.2: Minto Mine Site Layout and Receiving Environment

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decommissioning (e.g., of the water treatment plant and potentially the camp). It represents the period when all closure measures are fully established, including final water conveyances. Water treatment during post-closure 2 will be by passive treatment only.

1.2 Background

The Minto Mine's WUL includes Water Quality Objectives (WQOs) for lower Minto Creek at water quality monitoring station W2 (Figure 1.3). These WQOs apply during the operational period and are anticipated to also apply during the active closure period and in post-closure 1. A separate set of WQOs will be developed for the post-closure 2 period. The Minto Mine's WUL requires a Reclamation and Closure Plan (RCP) that includes WQOs for post-closure that meet the criteria for non-degradation of background concentrations. Two approaches have been provided in the WUL as guidance to developing closure WQOs (YWB 2015; Clause 112):

1) Establish closure WQOs that meet the criteria for non-degradation of background concentrations in Minto and McGinty creeks; and

2) If non-degradation of water quality is not achievable, the closure WQOs shall be set at no more than 50% of the assimilative capacity of Minto Creek using the following formula:

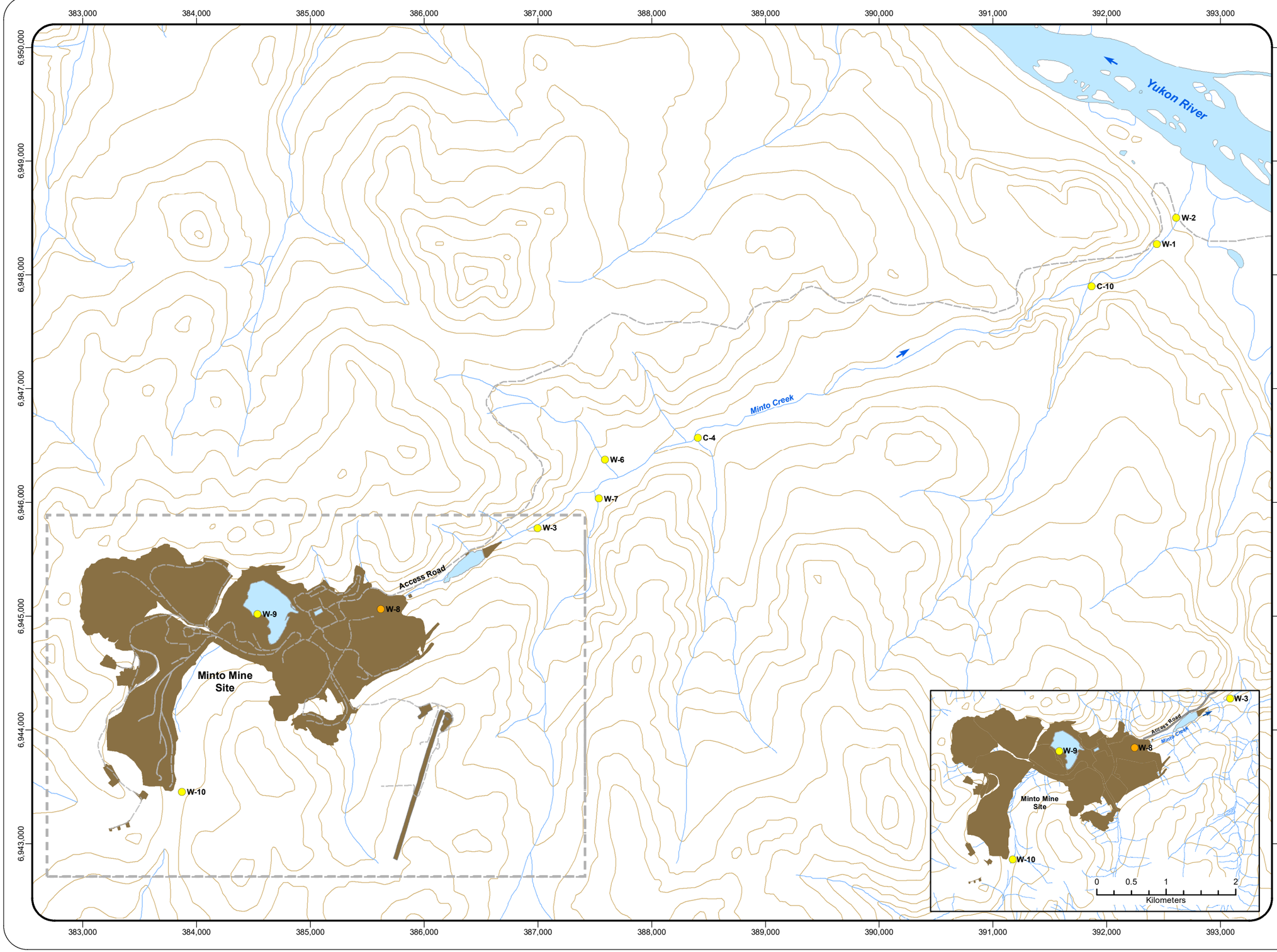
$$Closure\ WQO = BKGD + (0.5*[WQO_{Operations} - BKGD])$$

where: BKGD = background concentration.

Both approaches for the development of closure WQOs require a clearly-defined background water quality condition.

1.3 Objective

The objective of this report is to define the background water quality data set to be used to develop background water quality for lower Minto Creek and to determine background summary statistics for key analytes to be used in calculating the post-closure WQOs. This document outlines the approach used in considering what data points should be included to define background water quality. This document incorporates the outcomes of discussions among the Minto Mine, SFN, and technical consultants (Appendix A). These discussions covered a number of factors critical to the definition of background water quality, including representative sampling stations, contaminants of potential concern (COPCs), dealing with less than method detection limit (MDL) results, dealing with replicate results, identifying outliers, evaluation of possible data groupings, and designation of summary statistics. The



Features

- Excluded Water Quality Sampling Station
- Water Quality Sampling Station
- Waterbody
- Mine Footprint (2013)
- Watercourse
- Mine Road
- Contour
- Water Flow Direction

MAP INFORMATION
Datum: NAD 83 Map Projection: UTM Zone 8V
Data Source: National Topographic Data Base (NTDB) compiled by Department of Natural Resources Canada at a scale of 1:50,000. All rights reserved.

Mine infrastructure data provided by Access Mining Consultants Inc.
Mine site contours derived from 2012 aerial imagery obtained from Challenger Geomatics.

Creation Date: June 2016
Project No. 167202.0062

Figure 1.3: Water Quality Sampling Stations, 2005 - 2013, used to Develop Background Concentrations

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background water quality data set is then used to calculate summary statistics for the COPCs for application as WQOs for the post-closure 2 period.

1.4 Report Organization

Section 2.0 provides a detailed overview of data organization and step-wise work-through the process of defining background water quality for the development of post-closure WQOs. Section 3.0 provides the proposed background water quality statistics. All references cited in the report are provided in Section 4.0.

2.0 DATA ORGANIZATION

Raw data from the Minto Mine's water quality database were received on May 31st, 2016. The database contained concentrations of water quality analytes measured throughout the open water season from 2005 to 2013. The database was sorted and organized as described below, to obtain a final water quality data set to be used to calculate the background concentrations of COPCs. For a few sampling dates, it was noted that water quality analyses were incomplete (e.g., incomplete dissolved metals in a multi-element scan). Minto Mine and Access Consulting Group were contacted to inquire about any missing data but older data sets were unavailable as they were collected by The North Air Group (Vancouver, BC).

2.1 Sampling Stations

The database included six sampling stations (W1, W2, W3, W8, W9 and W10) that were sampled during baseline studies in 2005 and were subsequently influenced by the mine (Figure 1.3; Appendix B). The database also included four reference sampling stations, W6 (2005 – 2013), W7 (2005 – 2013), C4 (2012, 2013) and C10 (2012, 2013; Figure 1.3; Appendix B).

Data collected at sampling station W8 were removed from the final background data set because water quality at station W8 was determined to poorly represent water quality at station W2 (lower Minto Creek; the point of WQO application). Data collected at W6 from 2009 to 2011 were not included, as these water quality samples were accidently collected at the Minto Creek main stem and not at the W6 tributary. These data would not represent background concentrations because the mine was operational from 2009 to 2011.

2.2 Contaminants of Potential Concern

The COPCs considered herein are those identified in the Water Use Licence (WUL; QZ14-031; YWB 2015). These COPCs are ammonia, pH (laboratory and field), nitrite, nitrate and the following dissolved metals; aluminum, arsenic, cadmium, chromium, copper, iron, lead, molybdenum, nickel, selenium, silver and zinc. In addition to the COPCs, several additional analytes are considered herein due to their importance as supporting parameters (e.g., parameters that may affect other WQOs). These include dissolved organic carbon (DOC), total and dissolved hardness, total suspended solids (TSS) and total dissolved solids (TDS).

2.3 Method Detection Limits

Concentrations that were reported as less than MDL and where the associated MDL exceeded the maximum detected concentration were removed from the data set. Such cases (and consequent removals) were observed for dissolved chromium (11% of samples) and dissolved

silver (24% of samples; Table 2.1). The remaining analytes reported as less than method detection limits (MDL) were substituted with half the MDL for calculations of summary statistics.

2.4 Replicate Samples

Some water samples were collected multiple times in one day at a given sampling station. To account for this, the average concentration of the analyte at a given station on the same day was used to represent the concentration for that day. The relative percent difference (RPD) of the minimum and maximum daily concentrations were calculated prior to calculating a daily mean concentration (Table 2.2). If the RPD was greater than 25% they were investigated further (Table 2.2).

Many analytes that had RPD greater than 25% were dissolved metals or TSS. Although there were instances where the RPD was greater than 25%, this is expected in multi-element scans. The investigation indicated no systemic problems (e.g., instances where all analytes had RPD greater than 25%), therefore all data were retained in the estimate of the daily mean. Regardless of RPDs between the minimum and maximum concentrations, the mean of concentrations measured on the same day were considered to best represent the concentration on that day.

2.5 Outliers

Outliers were evaluated based on data distributions for each analyte. The term “outlier” in the context of this report is used to refer to an observation that is unusually small or large relative to the distribution of observations. Outliers were investigated further to determine whether the value was naturally possible or whether the value could have resulted from an analytical error. The background data distributions for all analytes were non-normally distributed. Outliers were therefore identified using a non-parametric rule (the lower outer fence and upper outer fence) that is frequently used to identify outliers on boxplots (Dawson 2011).

Lower outer fence = Q1 – 3 (IQR)
Upper outer fence = Q3 + 3 (IQR)
where: Q1 = 1st Quartile
Q3 = 3rd Quartile
IQR = Interquartile Range (Q3 – Q1)

Observations less than the lower outer fence or greater than the upper outer fence were considered outliers. Although the rule is non-parametric, the calculation of the IQR is influenced by data transformations. The outlier evaluation was therefore conducted using untransformed and $\log_{10}$-transformed concentrations because many distributions of water

Table 2.1: Samples with high Method Detection Limits (MDL) that were removed from the background dataset.

Analyte	Max Value (All Stations)	MDL greater than Max Value	Station	Total Sample Size	Number of samples that exceeded Max Value
Dissolved Chromium	0.0017	0.0020	W7	110	20
Dissolved Silver	0.000060	0.00010	W6	23	1
			W7	110	43

Table 2.2: Relative Percent Difference (RPD) of maximum and minimum values for samples collected at the same sampling stations and date. Highlighted cells had RPD > 25%.

Station	Date	DOC (mg/L)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	TSS (mg/L)	TDS (mg/L)	Ammonia (mg/L)	Nitrite (mg/L)	Nitrate (mg/L)	Lab pH (pH Units)	Field pH (pH Units)	Dissolved Aluminum (mg/L)	Dissolved Arsenic (mg/L)	Dissolved Cadmium (mg/L)	Dissolved Chromium (mg/L)	Dissolved Copper (mg/L)	Dissolved Iron (mg/L)	Dissolved Lead (mg/L)	Dissolved Molybdenum (mg/L)	Dissolved Nickle (mg/L)	Dissolved Selenium (mg/L)	Dissolved Silver (mg/L)	Dissolved Zinc (mg/L)
W2	5/27/2005			100	21	138	0.026	0.0015	0.057	8.06		0.016	0.00025	0.000025	0.00050	0.0033	0.16	0.00025	0.00050	0.0015	0.00050	0.000010	
				72	79	80	0.010	0.00050	0.034	8.03		0.011											0.0025
RPD				**32%**	**118%**	**53%**	**89%**	**100%**	**50%**	**0%**		**41%**											
W10	5/27/2005			46	1.5	115	0.027	0.00050	0.0094	7.47		0.14	0.00025	0.000025	0.00050	0.018	0.11	0.00025	0.00050	0.0016	0.00050	0.000010	0.0085
							0.010	0.00050					0.00025	0.000025	0.00050	0.00050	0.015	0.00025	0.00050	0.00050	0.00050	0.000010	0.0025
RPD							**92%**	**0%**					**0%**	**0%**	**0%**	**189%**	**150%**	**0%**	**0%**	**105%**	**0%**	**0%**	**109%**
W7	6/15/2006						0.027																
			100		53	114			0.18	7.86		0.013	0.00040	0.0000050	0.00025	0.0010	0.090	0.000050	0.0010	0.0010	0.00020		0.00050
RPD																							
W7	7/26/2006						0.016																
			128		9.0				0.18	7.87		0.0080	0.00040	0.0000050	0.00025	0.0020	0.10	0.000050	0.00050	0.00210	0.00010		0.00050
RPD																							
W7	8/25/2006						0.027																
			129			180			0.12	7.92		0.012	0.00040	0.0000050	0.00060	0.0010	0.080	0.000050	0.0010	0.00090	0.00010		0.0030
RPD																							
W7	6/5/2007			108	76	162	0.025	0.025	0.050	7.98													
				120	76	158	0.025	0.050	0.050	7.97													
				111	73	166	0.025	0.025	0.050	7.97													
RPD				**11%**	**4%**	**5%**	**0%**	**67%**	**0%**	**0%**													
W7	7/18/2007	14.3	128	126	4.0	170	0.025	0.025	0.30	7.72		0.011	0.00050	0.0000050	0.0010	0.0020	0.040	0.000050	0.0010	0.00070	0.00010		0.0060
				125	7.0	182	0.025	0.025	0.30	7.80													
				125	6.0	184	0.025	0.025	0.30	7.83													
RPD				**1%**	**55%**	**8%**	**0%**	**0%**	**0%**	**1%**													
W7	8/24/2007						0.025																
							0.025																
							0.025																
				122	4.0	198		0.025	0.20	8.04													
				123	4.0	192		0.025	0.20	8.05													
				128	6.0	186		0.025	0.20	8.04													
RPD				**5%**	**0%**	**6%**		**0%**	**0%**	**0%**													
W7	10/30/2007			130	3.0	176	0.025	0.14	0.010	7.74													
				128	10	182	0.025	0.14	0.010	7.74													
				130	7.0	166	0.025	0.13	0.010	7.74													
					6.0	172		0.13	0.010	7.68													
RPD				**2%**	**108%**	**9%**	**0%**	**7%**	**0%**	**1%**													
W7	4/22/2008		186		1.0	264	0.025	0.070	0.010	8.40		0.010	0.00010	0.000040	0.0012	0.00050	0.010	0.000050	0.0016	0.00050	0.00030		0.0010
			193		1.0	274	0.025	0.040	0.010	8.41		0.010	0.00040	0.000040	0.0014	0.0020	0.070	0.00010	0.0017	0.0020	0.00030		0.0030
			182		1.0	270	0.025	0.070	0.010	8.40		0.010	0.00030	0.000040	0.00030	0.0010	0.010	0.000050	0.0017	0.00050	0.00030		0.0010
					1.0	262		0.040	0.010	8.40													
RPD			**6%**		**0%**	**4%**	**0%**	**55%**	**0%**	**0%**		**0%**	**120%**	**0%**	**129%**	**120%**	**150%**	**67%**	**5%**	**120%**	**0%**		**100%**
W7	6/3/2008			97	1.0	134	0.025	0.040	0.030	8.01													
				96	19	132	0.025	0.040	0.040	8.01													
				101	26	146	0.025	0.040	0.040	8.03													
			92									0.005	0.00010	0.000040	0.0013	0.00050	0.050	0.000050	0.0012	0.0010	0.00030	0.0000050	0.0020
RPD				**5%**	**185%**	**10%**	**0%**	**0%**	**29%**	**0%**													

Table 2.2: Relative Percent Difference (RPD) of maximum and minimum values for samples collected at the same sampling stations and date. Highlighted cells had RPD > 25%.

Station	Date	DOC (mg/L)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	TSS (mg/L)	TDS (mg/L)	Ammonia (mg/L)	Nitrite (mg/L)	Nitrate (mg/L)	Lab pH (pH Units)	Field pH (pH Units)	Dissolved Aluminum (mg/L)	Dissolved Arsenic (mg/L)	Dissolved Cadmium (mg/L)	Dissolved Chromium (mg/L)	Dissolved Copper (mg/L)	Dissolved Iron (mg/L)	Dissolved Lead (mg/L)	Dissolved Molybdenum (mg/L)	Dissolved Nickle (mg/L)	Dissolved Selenium (mg/L)	Dissolved Silver (mg/L)	Dissolved Zinc (mg/L)
W7	8/6/2008			123	92	162	0.025	0.0050	0.14	7.9													
				123	139	168	0.025	0.0050	0.14	7.9													
				125	85	170		0.0050	0.15	8.0													
		9.0	123				0.025					0.014	0.00040	0.0000050	0.00060	0.0020	0.070	0.000050	0.0010	0.0010	0.00010	0.000060	0.0060
RPD				**2%**	**48%**	**5%**	**0%**	**0%**	**7%**	**1%**													
W7	10/28/2008				4.0	192	0.025			7.84													
					6.0	188	0.025			7.83													
					7.0	168	0.025			7.85													
		11	123									0.0090	0.00050	0.0000050	0.00050	0.0010	0.078	0.000050	0.0011	0.0010	0.00030	0.0000050	0.0010
RPD					**55%**	**13%**	**0%**			**0%**													
W7	10/7/2009																						
		6.9	123		7.0	180			0.12	7.87	8.24												
RPD																							
W7	10/25/2009	8.7	126	141	2.0	130	0.010	0.0025	0.15	8.10	8.23	0.0070	0.00030	0.000040	0.00050	0.0012	0.046	0.00010	0.0010	0.00050	0.00020	0.000010	0.0025
					2.0	110				8.00													
RPD					**0%**	**17%**				**1%**													
W7	9/12/2010			105																			
		17	107	102	11	210	0.039	0.0025	0.030	7.98	8.02	0.016	0.00050	0.000050		0.0010	0.23	0.00010	0.0010	0.0020	0.00040		0.0050
RPD				**3%**																			
W7	5/6/2012	17	46	45	2.4	80	0.010	0.0025	0.010	6.05		0.030	0.00043	0.000013	0.00050	0.0030	0.43	0.00010	0.00050	0.0011	0.000050	0.000010	0.0025
		17	46	45	4.2	88	0.011	0.0025	0.010	5.74	7.10	0.044	0.00044	0.000016	0.00050	0.0027	0.46	0.00010	0.00050	0.0012	0.000050	0.000010	0.0025
RPD		**2%**	**1%**	**0%**	**55%**	**10%**	**10%**	**0%**	**0%**	**5%**		**39%**	**2%**	**21%**	**0%**	**10%**	**8%**	**0%**	**0%**	**9%**	**0%**	**0%**	**0%**
W7	6/19/2013	11	113	114	11	152	0.052	0.0025	0.14	8.08		0.013	0.00052	0.0000050	0.00050	0.0012	0.29	0.00010	0.0015	0.0013	0.00011	0.000010	0.0025
		11	114	115	10	164	0.040	0.0025	0.14	8.07	8.60	0.013	0.00051	0.0000050	0.00050	0.0012	0.29	0.00010	0.0016	0.0012	0.00019	0.000010	0.0025
RPD		**7%**	**1%**	**1%**	**10%**	**8%**	**26%**	**0%**	**1%**	**0%**		**7%**	**2%**	**0%**	**0%**	**7%**	**2%**	**0%**	**6%**	**8%**	**53%**	**0%**	**0%**
C4	10/18/2013	13	129	142	16	214	0.091	0.0025	0.069	8.18		0.022	0.001060	0.000005	0.00050	0.0011	1.2	0.00010	0.00050	0.0023	0.000050	0.000010	0.0025
		13	129	139	16	220	0.080	0.0025	0.065	8.24	7.16	0.025	0.001070	0.000005	0.00050	0.0012	1.2	0.00010	0.0012	0.0025	0.000050	0.000010	0.0025
RPD		**2%**	**0%**	**2%**	**0%**	**3%**	**13%**	**0%**	**6%**	**1%**		**12%**	**1%**	**0%**	**0%**	**5%**	**1%**	**0%**	**82%**	**8%**	**0%**	**0%**	**0%**
C10	9/8/2013	16	150	163	25	186	0.046	0.0076	0.18	7.94		0.025	0.001660	0.0000050	0.00050	0.0013	1.3	0.00010	0.00050	0.0025	0.000050	0.000010	0.0025
		16	150	162	22	206	0.055	0.0070	0.19	8.13	8.16	0.024	0.001580	0.000010	0.0017	0.0013	1.3	0.00010	0.00050	0.0025	0.000050	0.000010	0.0025
RPD		**1%**	**0%**	**1%**	**16%**	**10%**	**18%**	**8%**	**9%**	**2%**		**5%**	**5%**	**67%**	**109%**	**6%**	**2%**	**0%**	**0%**	**0%**	**0%**	**0%**	**0%**

quality analytes are approximately log-normal. Since pH is reported on the logarithmic scale, the outlier rule was applied to the pH data that were untransformed and anti-logged. Any observations that were identified as potential outliers based on both the untransformed and log_{10}-transformed scale (Appendix C) were investigated further. When Q3 could not be estimated because it was below the MDL, the upper outer fence was estimated using the following formula:

$$MDL_{Q3} + 3 * MDL_{Q3}$$

where: MDL_{Q3} *= the MDL corresponding to the observation at the third quartile*

One observation - dissolved arsenic at station C4 on November 17th 2013 (0.0056 mg/L) - was identified as an outlier based on the above rules (Table 2.3). Further investigation of this potential outlier included consideration of potential elevations of other analytes and consideration of analyte ratios (i.e., chemical signature). Hardness, dissolved iron and dissolved manganese were also elevated in this sample. Because the potential outlying value appeared to be naturally possible, was consistent with observations of elevation of other analytes, and there was no unequivocal evidence of unusual ratios of arsenic relative to other analytes (Appendix Table D.1), the concentration of dissolved arsenic at station C4 was retained in the background data set.

2.6 Baseline Investigation

Water samples collected in 2005 before mine operations were considered baseline data. Concentrations of COPCs were plotted using an individual value plot for all sampling stations to determine if concentrations for each station were similar to station W2 and could be used as baseline data to represent station W2 (Figure 2.1). The median and 95th percentile of the background data sets for each analyte were calculated separately for W2 and all sampling stations (including W2) in 2005 (Table 2.4). There appears to be close agreement between the median and 95th percentiles for the two data sets (i.e., W2 and all sampling stations in 2005; Table 2.4). Total hardness differed most among data sets (was greater at station W2), but summary statistics for many analytes did not differ between W2 and all sampling stations (Table 2.4). In general, the distribution of concentrations of COPCs at station W2 was similar to the distribution of concentrations of COPCs from the other sampling stations in 2005, supporting the use of the designated reference water quality data set as representative of station W2.

Table 2.3: Outliers identified and rationale for removing or keeping the data point.

Station	Date	Identified Outlier			
		Analyte	Value	Decision	Rationale
C4	11/17/2013	Dissolved Arsenic (mg/L)	0.0056	Retained	High Hardness, dissolved iron and dissolved manganese in sample.

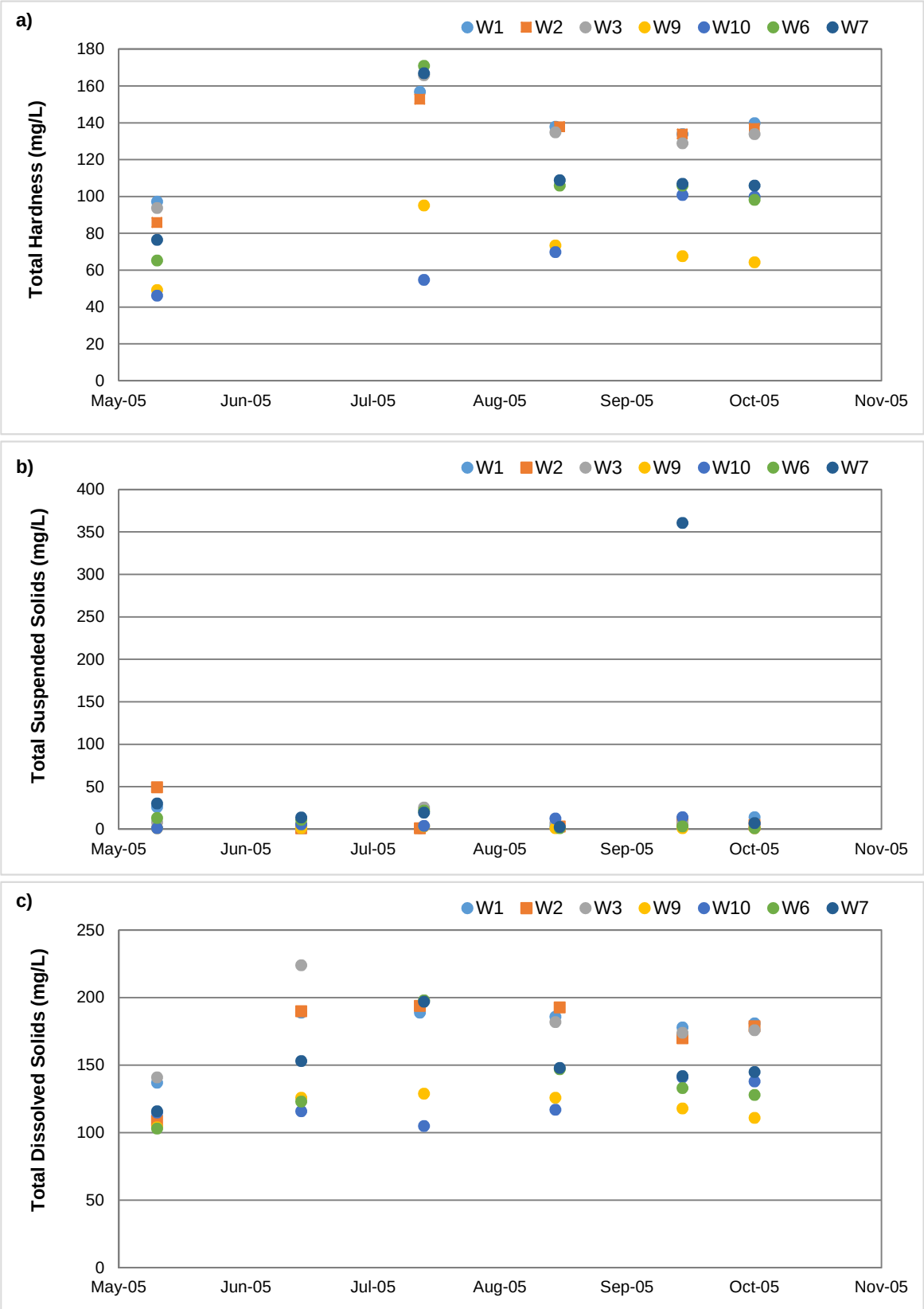


Figure 2.1: Baseline concentrations of contaminants of potential concern, Minto Mine, 2005. Values less than method detection limits were substituted with half of the MDL value.

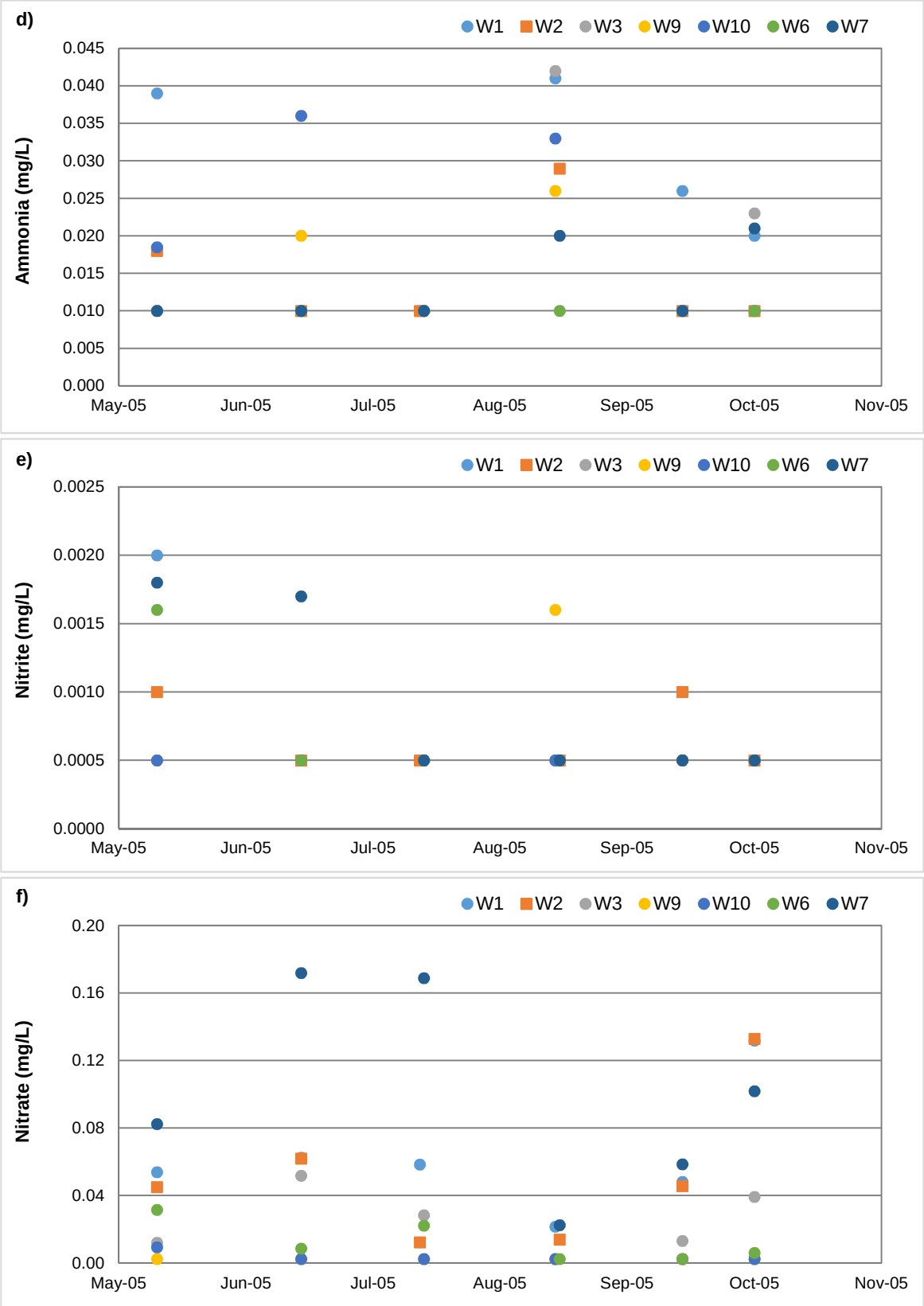


Figure 2.1: Baseline concentrations of contaminants of potential concern, Minto Mine, 2005. Values less than method detection limits were substituted with half of the MDL value.

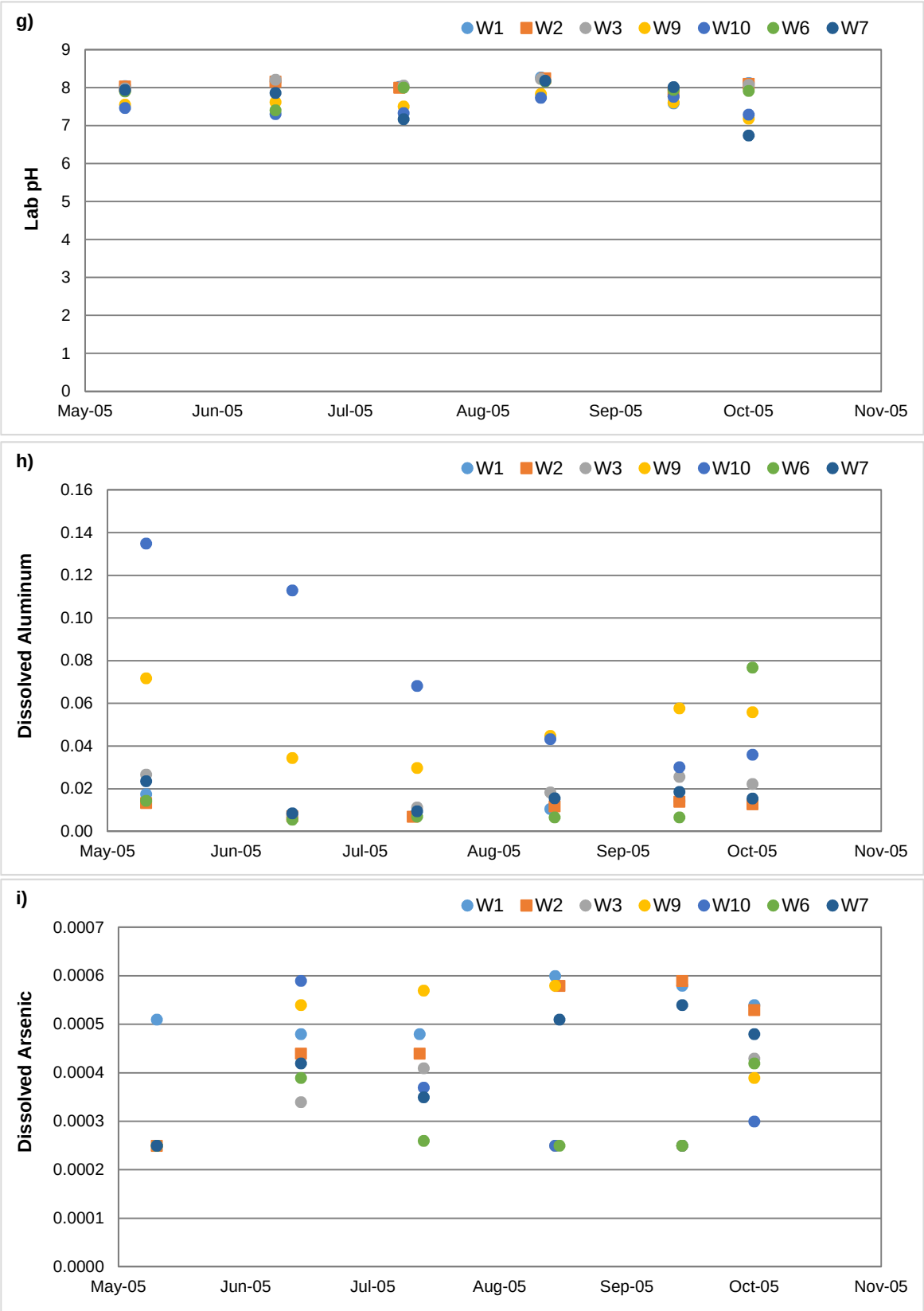


Figure 2.1: Baseline concentrations of contaminants of potential concern, Minto Mine, 2005. Values less than method detection limits were substituted with half of the MDL value.

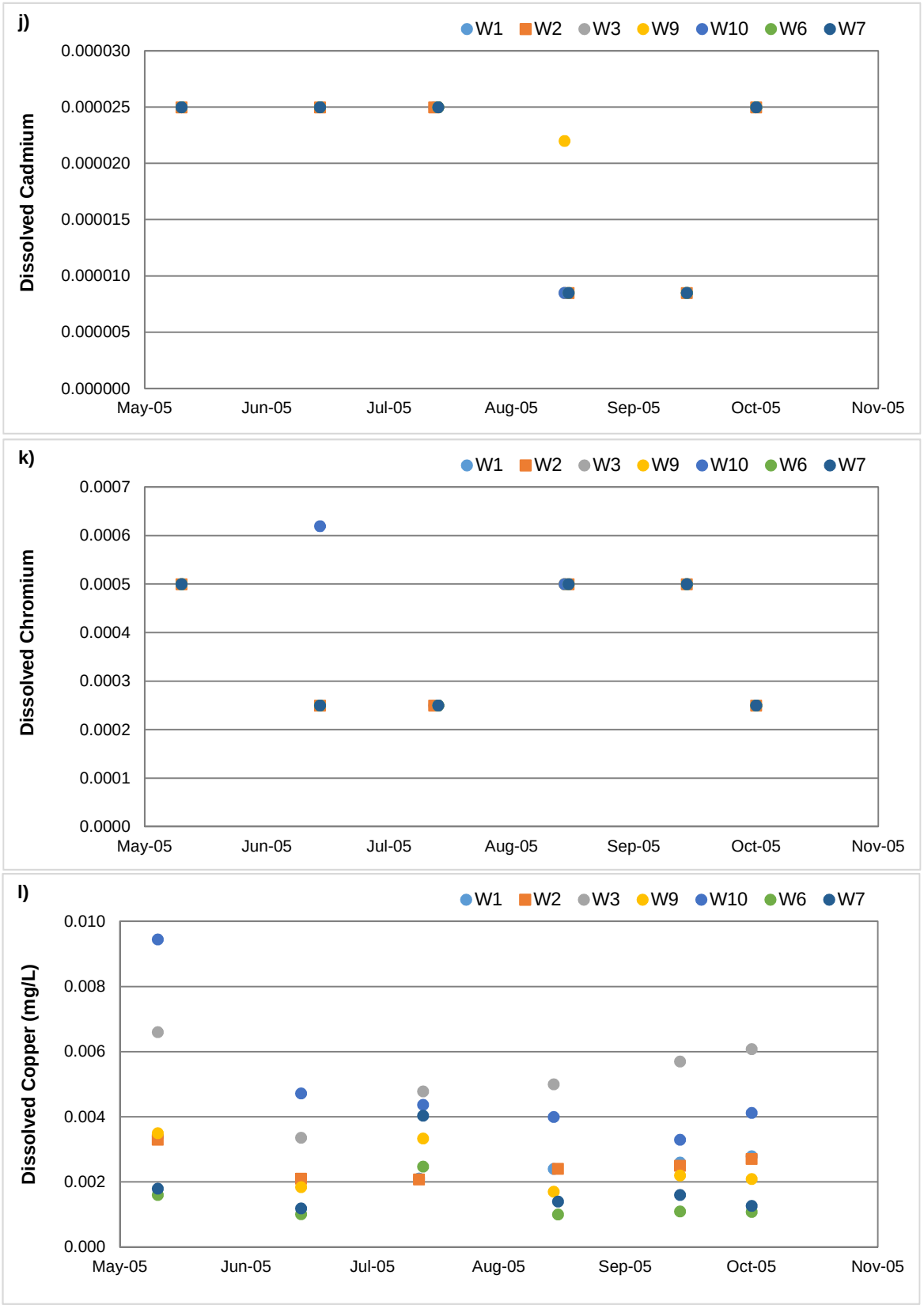


Figure 2.1: Baseline concentrations of contaminants of potential concern, Minto Mine, 2005. Values less than method detection limits were substituted with half of the MDL value.

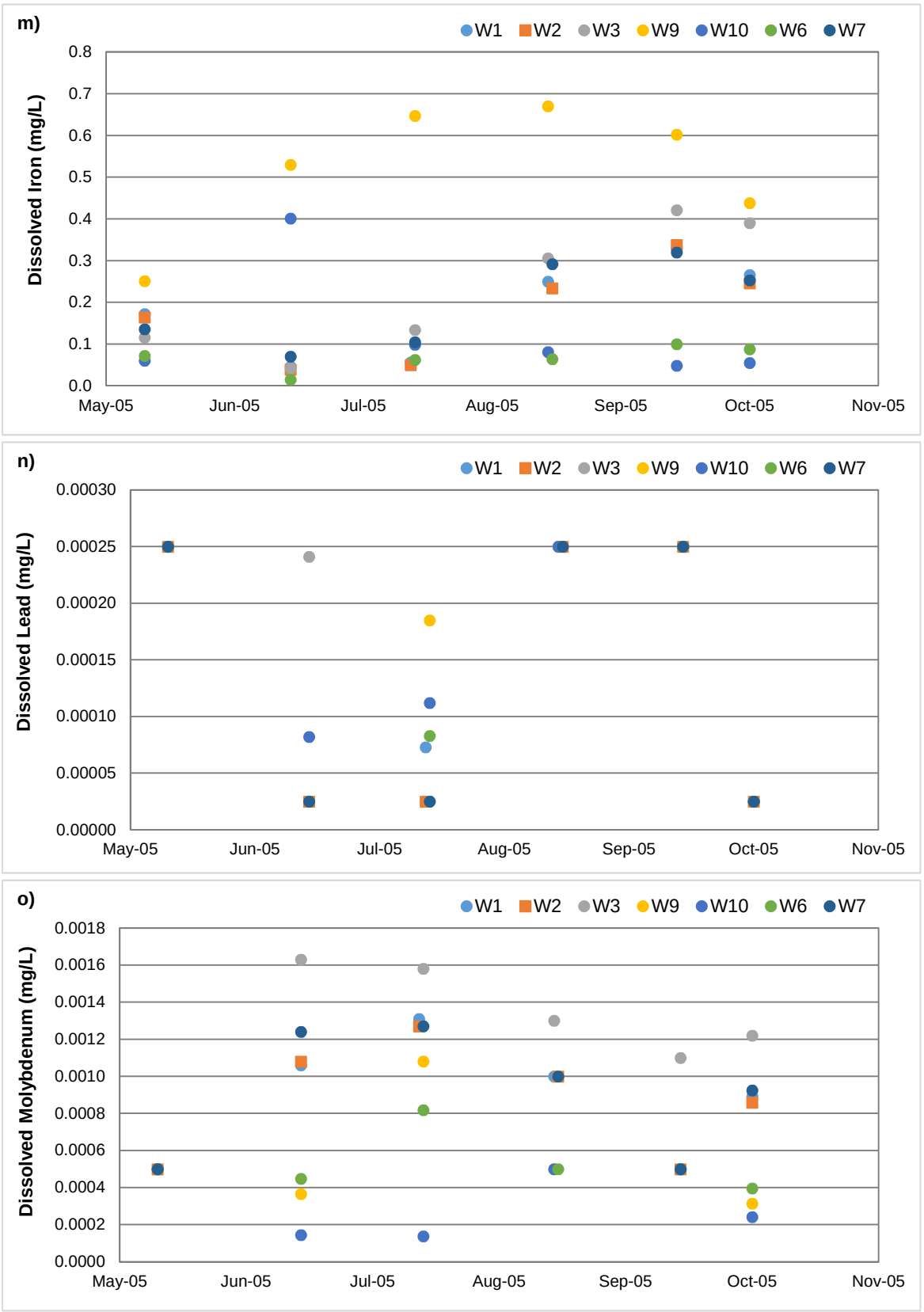


Figure 2.1: Baseline concentrations of contaminants of potential concern, Minto Mine, 2005. Values less than method detection limits were substituted with half of the MDL value.

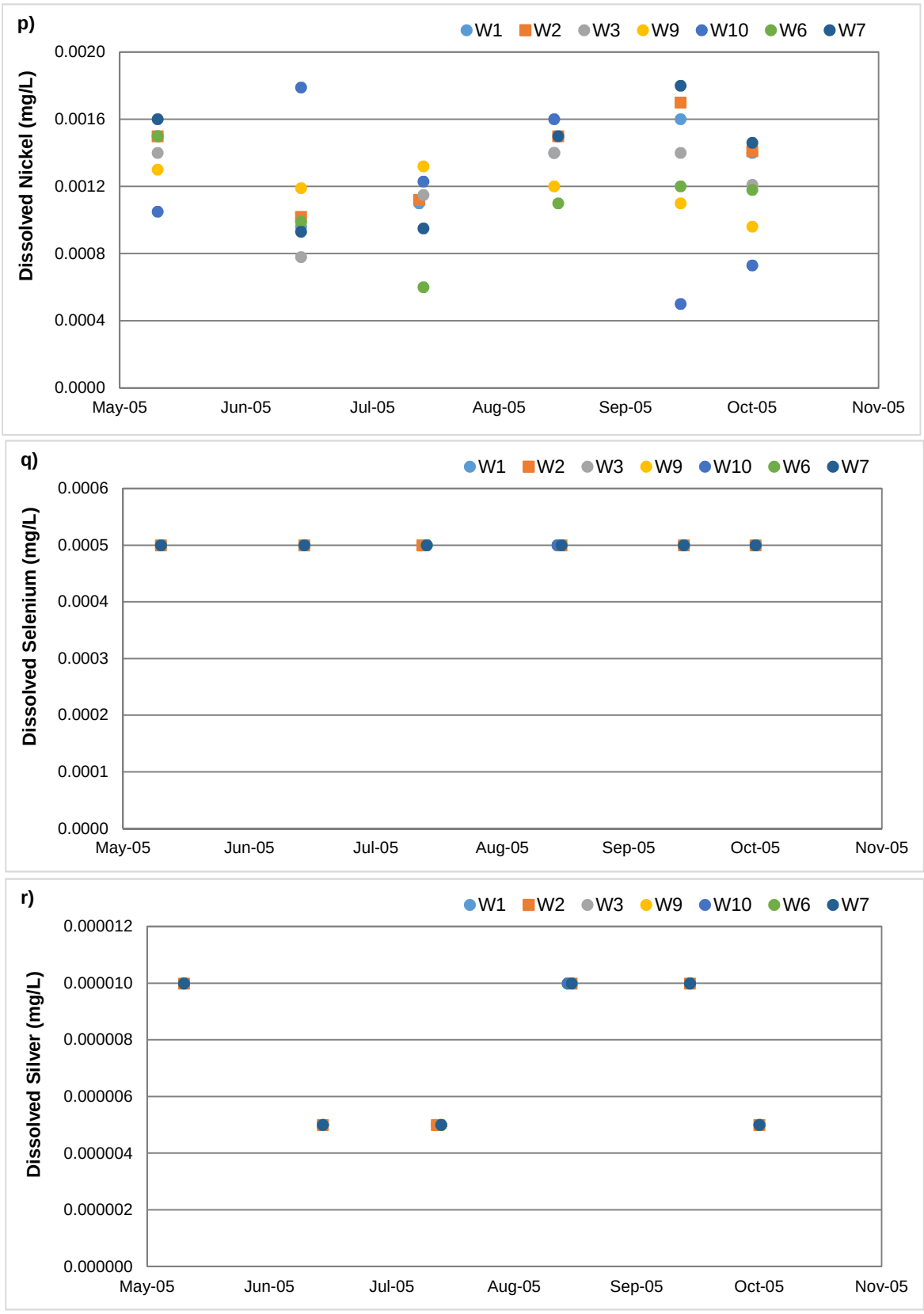


Figure 2.1: Baseline concentrations of contaminants of potential concern, Minto Mine, 2005. Values less than method detection limits were substituted with half of the MDL value.

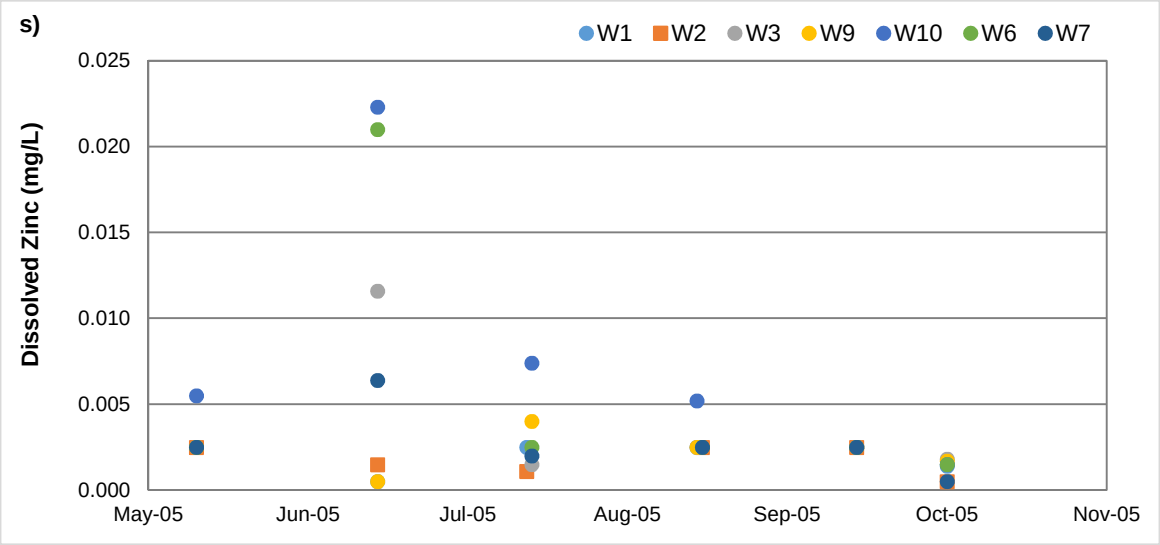


Figure 2.1: Baseline concentrations of contaminants of potential concern, Minto Mine, 2005. Values less than method detection limits were substituted with half of the MDL value.

Table 2.4: Median and 95th Percentile for W2 and All sampling stations (including W2) in 2005.

Analyte	W2			All Sampling Stations		
	Sample Size	Median	95th Percentile	Sample Size	Median	95th Percentile
Total Hardness (mg/L)	5	137	150	35	106	166
TSS (mg/L)	6	4.5	40	42	5.8	30
TDS (mg/L)	6	185	194	42	144	198
Ammonia (mg/L)	6	0.010	0.026	42	0.010	0.039
Nitrite (mg/L)	6	0.00050	0.0010	42	0.00050	0.0017
Nitrate (mg/L)	6	0.046	0.12	42	0.018	0.13
Lab pH (pH Units)	6	8.08	8.24	42	7.95	8.24
Dissolved Aluminum (mg/L)	6	0.012	0.014	42	0.016	0.077
Dissolved Arsenic (mg/L)	6	0.00049	0.00059	42	0.00042	0.00059
Dissolved Cadmium (mg/L)	6	0.000025	0.000025	42	0.000025	0.000025
Dissolved Chromium (mg/L)	6	0.00038	0.00050	42	0.00050	0.00050
Dissolved Copper (mg/L)	6	0.0025	0.0032	42	0.0025	0.0061
Dissolved Iron (mg/L)	6	0.20	0.32	42	0.15	0.60
Dissolved Lead (mg/L)	6	0.00014	0.00025	42	0.00025	0.00025
Dissolved Molybdenum (mg/L)	6	0.0009295	0.0012	42	0.00050	0.0013
Dissolved Nickel (mg/L)	6	0.0015	0.0017	42	0.0012	0.0017
Dissolved Selenium (mg/L)	6	0.00050	0.00050	42	0.00050	0.00050
Dissolved Silver (mg/L)	6	0.0000075	0.000010	42	0.0000075	0.000010
Dissolved Zinc (mg/L)	6	0.0020	0.0025	42	0.0025	0.011

2.7 Data Grouping

Water quality sampling was typically conducted in batches or “sampling campaigns” such that water quality samples were collected from multiple sampling stations on the same day or within a one week period. Samples collected on the same day may not be fully independent of each other, and therefore different methods of grouping the data were explored. Three options were evaluated: Option 1 did not involve any grouping (i.e., all data were used), Option 2 grouped samples by week, and Option 3 grouped samples by month (Table 2.5; Figure 2.2; Appendix Table D.1). Data were combined for Option 2 by calculating the mean from all sampling stations that were collected within a one week period (Table D.2). Data were combined for Option 3 in two steps: 1) by calculating the mean for each individual sampling station by month; and 2) then calculating the mean of the monthly sampling station means (Table D.3).

2.8 Summary Statistics

The designated reference water quality data set, developed as outlined above, was used to calculate summary statistics for each of the three options. For each analyte, the sample size, mean, median and 95th percentile were calculated for all stations combined over time for all three Options (Table 2.6). In addition, annual medians were calculated for each of the three options, and the 95th percentile of those annual medians was calculated (Table 2.7).

Table 2.5: Data grouping decisions for Option 2 and Option 3.

Year	Month	Day	Sampling Station									Option 2	Option 3
			W1	W2	W3	W9	W10	W6	W7	C4	C10		
2005	May	27	X	X	X	X	X	X	X	-	-	• Average of all sampling stations	-
	June	30	X	X	X	X	X	X	X	-	-	• Average of all sampling stations	-
	July	28	X	X	-	-	-	-	-	-	-	• Average of all sampling stations from both days	-
		29	-	-	X	X	X	X	X	-	-		
	August	29	X	-	X	X	X	-	-	-	-	• Average of all sampling stations from both days	-
		30	-	X	-	-	-	X	X	-	-		
	September	28	X	X	X	X	X	X	X	-	-	• Average of all sampling stations	-
	October	15	X	X	X	X	X	X	X	-	-	• Average of all sampling stations	-
2006	June	2	-	-	-	-	-	X	X	-	-	• Average of all sampling stations	• Average of W7 sampling stations • Take average of W7 (averaged value) and W6
		8	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
		15	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
		23	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
		28	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
	July	7	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	• Average of W7 sampling stations
		12	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
		20	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
		26	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
	August	2	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	• Average of W7 sampling stations • Take average of W7 (averaged value) and W6
		10	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
		25	-	-	-	-	-	X	X	-	-	• Average of all sampling stations	
		30	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
	September	6	-	-	-	-	-	X	X	-	-	• Average of all sampling stations	• Average of W7 sampling stations • Take average of W7 (averaged value) and W6
		13	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
		20	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
		28	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
	October	4	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	• Average of W7 sampling stations
		12	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
2007	June	5	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	• Average of W7 sampling stations • Take average of W7 (averaged value) and W6
		20	-	-	-	-	-	X	X	-	-	• Average of all sampling stations	
	July	18	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	August	24	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	September	18	-	-	-	-	-	X	X	-	-	• Average of all sampling stations	-
	October	30	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
2008	April	22	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	June	3	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	• Average of W7 sampling stations • Take average of W7 (averaged value) and W6
		17	-	-	-	-	-	X	X	-	-	• Average of all sampling stations	
	August	6	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	September	18	-	-	-	-	-	X	X	-	-	• Average of all sampling stations	-
	October	28	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
2009	April	30	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	June	8	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	July	28	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	August	6	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	• Average of W7 sampling stations
		25	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
		31	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
	October	7	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	• Average of W7 sampling stations
		25	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	

Table 2.5: Data grouping decisions for Option 2 and Option 3.

Year	Month	Day	Sampling Station									Option 2	Option 3
			W1	W2	W3	W9	W10	W6	W7	C4	C10		
2010	April	19	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	• Average of W7 sampling stations
		27	-	-	-	-	-	-	X	-	-	• Average of W7 sampling stations on both days	
		29	-	-	-	-	-	-	X	-	-		
	May	1	-	-	-	-	-	-	X	-	-	• Average of W7 sampling stations on all four days	• Average of W7 sampling stations
		3	-	-	-	-	-	-	X	-	-		
		5	-	-	-	-	-	-	X	-	-		
		7	-	-	-	-	-	-	X	-	-		
		9	-	-	-	-	-	-	X	-	-	• Average of W7 sampling stations on all four days	
		10	-	-	-	-	-	-	X	-	-		
		12	-	-	-	-	-	-	X	-	-		
		14	-	-	-	-	-	-	X	-	-		
		16	-	-	-	-	-	-	X	-	-	• Average of W7 sampling stations on all four days	
		17	-	-	-	-	-	-	X	-	-		
		19	-	-	-	-	-	-	X	-	-		
		21	-	-	-	-	-	-	X	-	-		
		23	-	-	-	-	-	-	X	-	-	• Average of W7 sampling stations on all four days	
		25	-	-	-	-	-	-	X	-	-		
		27	-	-	-	-	-	-	X	-	-		
		29	-	-	-	-	-	-	X	-	-		
		31	-	-	-	-	-	-	X	-	-	• Average of W7 sampling stations on all four days	
	June	2	-	-	-	-	-	-	X	-	-		• Average of W7 sampling stations
		4	-	-	-	-	-	-	X	-	-		
		6	-	-	-	-	-	-	X	-	-		
		14	-	-	-	-	-	-	X	-	-	• Average of W7 sampling stations on all four days	
		16	-	-	-	-	-	-	X	-	-		
		18	-	-	-	-	-	-	X	-	-		
		20	-	-	-	-	-	-	X	-	-		
		22	-	-	-	-	-	-	X	-	-	• Average of W7 sampling stations on all four days	
		24	-	-	-	-	-	-	X	-	-		
		26	-	-	-	-	-	-	X	-	-		
		28	-	-	-	-	-	-	X	-	-		
	July	5	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	August	5	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	September	12	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	October	8	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
2011	April	27	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	May	2	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	June	1	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	• Average of W7 sampling stations
		16	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	
	July	1	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	August	2	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	• Average of W7 sampling stations
		16	-	-	-	-	-	-	X	-	-	• Average of W7 sampling stations	
		19	-	-	-	-	-	-	X	-	-		
	September	12	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	October	6	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	November	22	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-

Table 2.5: Data grouping decisions for Option 2 and Option 3.

Year	Month	Day	Sampling Station									Option 2	Option 3
			W1	W2	W3	W9	W10	W6	W7	C4	C10		
2012	January	24	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	February	25	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	March	5	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	April	24	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	May	6	-	-	-	-	-	-	X	X	-	• Average of all sampling stations from both days	-
		12	-	-	-	-	-	-	-	-	X		
	June	19	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	July	17	-	-	-	-	-	X	X	-	-	• Average of all sampling stations	• Average sampling stations separately • Using station averages take average for all days
		30	-	-	-	-	-	X	X	-	X	• Average of all sampling stations	
	August	1	-	-	-	-	-	-	-	X	-	• Use C4 sampling station	• Average sampling stations separately • Using station averages take average for all days
		8	-	-	-	-	-	-	X	X	-	• Average of all sampling stations from all days	
		9	-	-	-	-	-	X	-	-	-		
		10	-	-	-	-	-	-	-	-	X		
		23	-	-	-	-	-	-	X	-	-	• Average of all sampling stations from all days	
		25	-	-	-	-	-	X	-	X	-		
		27	-	-	-	-	-	-	-	-	X		
	September	13	-	-	-	-	-	X	X	-	-	• Average of all sampling stations	-
	October	13	-	-	-	-	-	X	-	-	-	• Average of all sampling stations from both days	-
		16	-	-	-	-	-	-	X	-	-		
	November	2	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	• Average of all sampling stations from both days
		12	-	-	-	-	-	X	-	-	-	• Use W6 sampling station	
	December	29	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
2013	April	30	-	-	-	-	-	-	X	-	-	• Use W7 sampling station	-
	May	14	-	-	-	-	-	X	X	-	-	• Average of all sampling stations	-
	June	19	-	-	-	-	-	-	X	-	-	• Average of all sampling stations from both days	-
		22	-	-	-	-	-	X	-	X	X		
	July	12	-	-	-	-	-	X	-	-	-	• Average of all sampling stations from both days	• Average of all sampling stations
		13	-	-	-	-	-	-	X	-	-		
		19	-	-	-	-	-	-	-	X	-	• Average of all sampling stations from both days	
		23	-	-	-	-	-	-	-	-	X		
	August	21	-	-	-	-	-	-	X	-	-	• Average of all sampling stations from both days	-
		22	-	-	-	-	-	X	-	X	X		
	September	6	-	-	-	-	-	X	X	X	-	• Average of all sampling stations from both days	-
		8	-	-	-	-	-	-	-	-	X		
	October	7	-	-	-	-	-	X	-	-	-	• Use W6 sampling station	• Average of all sampling stations
		18	-	-	-	-	-	-	-	X	X	• Average of all sampling stations from both days	
		21	-	-	-	-	-	-	X	-	-		
	November	16	-	-	-	-	-	X	X	-	-	• Average of all sampling stations from both days	-
		17	-	-	-	-	-	-	-	X	-		
	December	7	-	-	-	-	-	X	X	-	-	• Average of all sampling stations	-

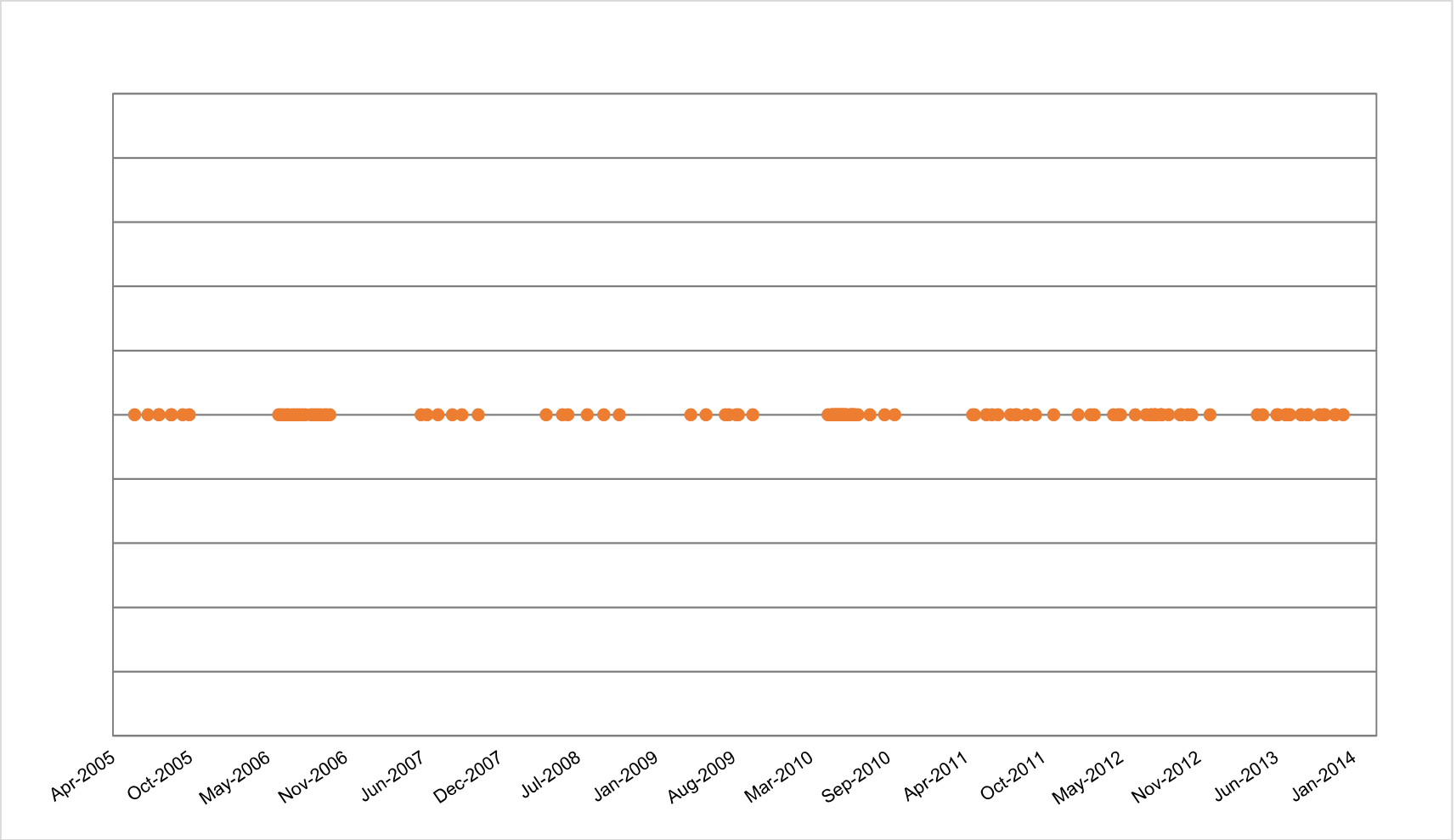


Figure 2.2: Distribution of water quality sampling events from 2005 to 2013.

Table 2.6: Summary statistics for Options 1, 2 and 3.

Analytes	Option 1 (No Grouping)				Option 2 (Grouped by Week)				Option 3 (Grouped by Month)			
	Sample Size	Mean	Median	95th Percentile	Sample Size	Mean	Median	95th Percentile	Sample Size	Mean	Median	95th Percentile
DOC (mg/L)	108	12	11	22	59	12	12	21	46	13	12	21
Dissolved Hardness (mg/L)	134	113	117	162	80	115	120	169	53	113	118	181
Total Hardness (mg/L)	149	120	121	200	66	120	121	185	49	120	124	185
TSS (mg/L)	184	114	11	772	89	89	11	485	61	77	11	278
TDS (mg/L)	178	153	150	217	83	156	158	216	62	158	160	226
Ammonia (mg/L)	175	0.040	0.021	0.15	83	0.036	0.025	0.12	56	0.037	0.025	0.12
Nitrite (mg/L)	161	0.0071	0.0025	0.025	67	0.010	0.0025	0.038	52	0.010	0.0025	0.040
Nitrate (mg/L)	181	0.085	0.067	0.21	85	0.10	0.090	0.22	57	0.091	0.075	0.22
Lab pH (pH Units)	187	7.93	8.00	8.26	91	7.92	7.97	8.20	61	7.90	7.96	8.18
Field pH (pH Units)	93	7.73	7.76	8.30	46	7.72	7.77	8.24	33	7.71	7.80	8.13
Dissolved Aluminum (mg/L)	179	0.026	0.016	0.071	88	0.025	0.014	0.056	59	0.023	0.019	0.053
Dissolved Arsenic (mg/L)	179	0.00057	0.00044	0.0015	88	0.00053	0.00041	0.0013	59	0.00053	0.00041	0.0011
Dissolved Cadmium (mg/L)	179	0.000019	0.0000085	0.000050	88	0.000020	0.0000053	0.000050	59	0.000019	0.000010	0.000050
Dissolved Chromium (mg/L)	159	0.00054	0.00050	0.0011	81	0.00060	0.00050	0.0013	55	0.00057	0.00050	0.0011
Dissolved Copper (mg/L)	179	0.0021	0.0016	0.0050	88	0.0021	0.0015	0.0051	59	0.0024	0.0017	0.0054
Dissolved Iron (mg/L)	179	0.38	0.11	1.2	88	0.31	0.11	0.95	59	0.34	0.15	0.91
Dissolved Lead (mg/L)	179	0.00012	0.00010	0.00025	88	0.00012	0.00010	0.00028	59	0.00012	0.00010	0.00025
Dissolved Molybdenum (mg/L)	179	0.00096	0.0010	0.0020	88	0.00097	0.0010	0.0017	59	0.00094	0.0010	0.0015
Dissolved Nickel (mg/L)	179	0.0013	0.0011	0.0027	88	0.0013	0.0011	0.0024	59	0.0012	0.0011	0.0020
Dissolved Selenium (mg/L)	179	0.00027	0.00020	0.00050	88	0.00023	0.00011	0.00050	59	0.00025	0.00015	0.00050
Dissolved Silver (mg/L)	137	0.0000092	0.000010	0.000010	60	0.0000098	0.000010	0.000010	47	0.000010	0.000010	0.000010
Dissolved Zinc (mg/L)	179	0.0032	0.0025	0.0060	88	0.0029	0.0025	0.0060	59	0.0033	0.0025	0.0084

Table 2.7: Annual Medians and the 95th percentile of annual medians for Option 1, 2 and 3.

Option	Year	Annual Medians																				
		DOC (mg/L)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	TSS (mg/L)	TDS (mg/L)	Ammonia (mg/L)	Nitrite (mg/L)	Nitrate (mg/L)	Lab pH (pH Units)	Field pH (pH Units)	Dissolved Aluminum (mg/L)	Dissolved Arsenic (mg/L)	Dissolved Cadmium (mg/L)	Dissolved Chromium (mg/L)	Dissolved Copper (mg/L)	Dissolved Iron (mg/L)	Dissolved Lead (mg/L)	Dissolved Molybdenum (mg/L)	Dissolved Nickel (mg/L)	Dissolved Selenium (mg/L)	Dissolved Silver (mg/L)	Dissolved Zinc (mg/L)
Option 1 (No Group)	2005	-	-	106	5.8	144	0.010	0.00050	0.018	7.95	-	0.016	0.00042	0.000025	0.00050	0.0025	0.15	0.00025	0.00050	0.0012	0.00050	0.0000075	0.0025
	2006	-	128	-	4.0	164	0.015	0.0010	0.11	7.85	-	0.0080	0.00042	0.0000050	0.00070	0.0010	0.090	0.000050	0.0010	0.00090	0.00010	0.0000050	0.0010
	2007	11	110	119	7.3	173	0.025	0.025	0.08	7.90	-	0.011	0.00040	0.0000050	0.0010	0.0020	0.050	0.000050	0.0010	0.00070	0.00010	-	0.0040
	2008	14	123	98	13	161	0.025	0.040	0.030	7.91	-	0.012	0.00030	0.000005	0.0013	0.0012	0.070	0.000050	0.0011	0.0010	0.00030	0.0000050	0.0020
	2009	8.7	121	141	2.8	148	0.010	0.0025	0.15	8.02	7.89	0.010	0.00030	0.000040	0.00050	0.0020	0.050	0.00010	0.0010	0.00050	0.00030	0.0000050	0.0025
	2010	9.2	99	106	18	140	0.022	0.0025	0.070	8.10	7.72	0.017	0.00040	0.000050	0.00050	0.0017	0.07	0.00010	0.0010	0.0010	0.00040	0.000010	0.0050
	2011	18	93	94	22	130	0.015	0.0025	0.055	7.96	7.79	0.033	0.00060	0.000010	0.00050	0.0019	0.35	0.00010	0.00050	0.0020	0.00010	0.000010	0.0025
	2012	12	129	145	45	166	0.025	0.0025	0.078	8.11	7.84	0.019	0.00078	0.0000050	0.00050	0.0016	0.58	0.00010	0.0011	0.0015	0.00012	0.000010	0.0025
	2013	11	129	153	23	176	0.049	0.0025	0.10	8.09	7.75	0.014	0.00059	0.0000050	0.00050	0.0012	0.36	0.00010	0.0011	0.0013	0.000050	0.000010	0.0025
	95th Percentile of Annual Medians	**17**	**129**	**150**	**36**	**175**	**0.040**	**0.034**	**0.13**	**8.11**	**7.88**	**0.027**	**0.00071**	**0.000046**	**0.0012**	**0.0023**	**0.49**	**0.00019**	**0.0011**	**0.0018**	**0.00046**	**0.000010**	**0.0046**
Option 2 (Grouped by Week)	2005	-	-	111	8.4	154	0.015	0.00061	0.040	7.83	-	0.025	0.00042	0.000025	0.00040	0.0028	0.21	0.00016	0.00076	0.0013	0.00050	0.0000075	0.0027
	2006	-	126	-	4.0	170	0.014	0.0010	0.12	7.85	-	0.0080	0.00040	0.0000050	0.00070	0.0010	0.090	0.000050	0.0010	0.00085	0.00010	0.0000050	0.0010
	2007	11	110	119	8.8	176	0.025	0.025	0.13	7.89	-	0.011	0.00040	0.0000050	0.0010	0.0020	0.050	0.000050	0.0010	0.00070	0.00010	-	0.0040
	2008	11	123	98	14	163	0.025	0.040	0.037	7.90	-	0.012	0.00031	0.000023	0.0011	0.0011	0.070	0.000050	0.0012	0.0010	0.00030	0.0000050	0.0020
	2009	8.7	121	141	2.8	148	0.010	0.0025	0.15	8.02	7.89	0.010	0.00030	0.000040	0.00050	0.0020	0.050	0.00010	0.0010	0.00050	0.00030	0.0000050	0.0025
	2010	12	106	106	13	145	0.027	0.0036	0.070	8.06	7.69	0.017	0.00033	0.000050	0.00050	0.0020	0.10	0.00010	0.0010	0.0010	0.00040	0.000010	0.0050
	2011	16	94	95	23	125	0.015	0.0025	0.050	7.98	7.79	0.029	0.00050	0.000010	0.00050	0.0019	0.28	0.00010	0.00050	0.0016	0.00010	0.000010	0.0025
	2012	11	129	154	30	164	0.025	0.0025	0.083	8.08	7.80	0.021	0.00060	0.0000050	0.00050	0.0016	0.50	0.00010	0.0011	0.0014	0.00012	0.000010	0.0025
	2013	12	129	151	34	179	0.078	0.0025	0.10	8.09	7.80	0.015	0.00088	0.0000050	0.00050	0.0012	0.59	0.00010	0.0010	0.0018	0.000093	0.000010	0.0025
	95th Percentile of Annual Medians	**15**	**129**	**153**	**33**	**178**	**0.058**	**0.034**	**0.14**	**8.09**	**7.87**	**0.027**	**0.00077**	**0.000046**	**0.0011**	**0.0025**	**0.55**	**0.00014**	**0.0011**	**0.0017**	**0.00046**	**0.000010**	**0.0046**
Option 3 (Grouped by Month)	2005	-	-	111	8.4	154	0.015	0.00061	0.038	7.83	-	0.025	0.00042	0.000025	0.00040	0.0028	0.21	0.00016	0.00076	0.0013	0.00050	0.0000075	0.0029
	2006	-	120	-	3.8	175	0.017	0.0010	0.083	7.78	-	0.0079	0.00046	0.0000050	0.00059	0.0011	0.10	0.000050	0.00088	0.00083	0.00013	0.0000050	0.0017
	2007	11	110	124	6.5	177	0.025	0.025	0.14	7.84	-	0.011	0.00040	0.0000050	0.0010	0.0020	0.050	0.000050	0.0010	0.00070	0.00010	-	0.0040
	2008	11	123	108	15	167	0.025	0.030	0.033	7.90	-	0.013	0.00035	0.0000050	0.0010	0.0012	0.070	0.000050	0.0011	0.0010	0.00030	0.0000050	0.0020
	2009	10	96	141	4.0	149	0.010	0.0025	0.16	7.90	7.73	0.019	0.00028	0.000035	0.00055	0.0028	0.060	0.00010	0.00092	0.00067	0.00030	0.0000050	0.0033
	2010	13	107	106	32	150	0.037	0.0031	0.070	8.04	7.85	0.016	0.00038	0.000050	0.00050	0.0020	0.13	0.00010	0.0010	0.0010	0.00040	0.000010	0.0050
	2011	19	79	89	17	125	0.016	0.0025	0.050	7.81	7.79	0.036	0.00050	0.000016	0.00050	0.0029	0.26	0.00010	0.00050	0.0012	0.00010	0.000010	0.000010
	2012	11	128	156	19	167	0.024	0.0025	0.092	8.10	7.59	0.021	0.00056	0.0000056	0.00050	0.0018	0.45	0.00010	0.0011	0.0014	0.00014	0.000010	0.0025
	2013	12	130	151	40	185	0.080	0.0037	0.11	8.09	7.90	0.019	0.0011	0.0000056	0.00050	0.0012	0.65	0.00010	0.0011	0.0018	0.000093	0.000010	0.0025
	95th Percentile of Annual Medians	**17**	**129**	**154**	**37**	**182**	**0.063**	**0.028**	**0.15**	**8.10**	**7.89**	**0.032**	**0.00086**	**0.000044**	**0.00099**	**0.0029**	**0.57**	**0.00014**	**0.0011**	**0.0016**	**0.00046**	**0.000010**	**0.0046**

3.0 BACKGROUND WATER QUALITY SUMMARY STATISTICS

Based primarily on consideration of potential water quality monitoring frequencies at the post-closure 2 stage (proposed spring, summer, fall), Option 3 is put forward as the proposed basis for defining background water quality. Two key sets of background water quality statistics are appropriate for defining background water quality:

1. the 95th percentile of the background data; and
2. the 95th percentile of the annual medians of the background data.

The intended use of the former (Table 3.1; Appendix Table D.4) is as an “individual data point evaluator” and the intended use of the latter (Table 3.1; Appendix Tables D.5 and D.6) is as a “central tendency evaluator”.

Although application of the WQOs in defining attainment is the subject of ongoing discussion between the Minto Mine and Selkirk First Nation, a fundamental guiding principle is that, because the summary statistics are 95th percentiles, they will be exceeded at an average rate of 5% in the absence of any mine influence on water quality. Thus, definition of attainment must consider natural exceedance rate, and initial response to any observed exceedance must be focussed on distinguishing a mine related influence from natural variability.

Table 3.1: Background water quality summary statistics representing lower Minto Creek (Minnow 2016) [1]

Analytes	Background Concentration Summary Statistics	
	Individual Data Point Evaluator (95th percentile)	Central Tendency Evaluator (95th percentile of annual medians)
Ammonia (mg/L)	0.12	0.063
Nitrite (mg/L)	0.040	0.028
Nitrate (mg/L)	0.22	0.15
Dissolved Aluminum (mg/L)	0.053	0.032
Dissolved Arsenic (mg/L)	0.0011	0.00086
Dissolved Cadmium (mg/L)	0.000050	0.000044
Dissolved Chromium (mg/L)	0.0011	0.0010
Dissolved Copper (mg/L)	0.0054	0.0029
Dissolved Iron (mg/L)	0.91	0.57
Dissolved Lead (mg/L)	0.00025	0.00014
Dissolved Molybdenum (mg/L)	0.0015	0.0011
Dissolved Nickel (mg/L)	0.0020	0.0016
Dissolved Selenium (mg/L)	0.00050	0.00046
Dissolved Silver (mg/L)	0.000010	0.000010
Dissolved Zinc (mg/L)	0.0084	0.0046

[1] for contaminants of potential concern (COPCs)

4.0 REFERENCES

Dawson, R. 2011. How significant is a boxplot outlier? Journal of Statistics Education 19: 2.

YWB (Yukon Water Board). 2015. Minto Mine Type A Water Use Licence Number QZ14-031. Issued to Minto Explorations Ltd. August 2015.

APPENDIX A

JUNE 2ND, 2016 MEETING MINUTES

Minto Mine Bilateral Technical Working Group Meeting
Characterizing Background Water Quality for Use as Water Quality Objectives
SRK Consulting – Vancouver, June 2, 2016
Meeting Notes

Attendees: Bill Slater, Don MacDonald, Jesse Sinclair, Ryan Herbert, Pierre Stecko, Dylan MacGregor

Agenda (developed jointly)

- Background Dataset
- Dealing with Duplicates
- Multiple locations per sampling date
- Dealing with Outliers
- Method Detection Limits
- Conditions / Binning
- Summary Statistics
- Attainment
- Confirmation of COPC list
- Water Quality Predictions Use

Key

Black text = items discussed
Blue text = agreed path forward
Red text = action item

1. Background Dataset

- Rationalize stations
- Test differences? Necessary?
- NOTE → W6 samples were taken at the wrong location in 2009-2011 (see extreme copper concentration from April 09) … exclude W6 for this timeframe
- Exclude W8
- "Background Dataset" was agreed upon … see PHOTOGRAPH of whiteboard matrix
- As with Access 2014, removed W6 data from 2009 to 2011 due to uncertainty about sampling location (may have been sampled in Minto Creek mainstem at some times)
- Prepare the dataset per spreadsheet
- Check against Access 2014. The only difference should be the exclusion of W8

2. Dealing with Duplicates

- There are duplicates in the database, some instances of greater replication
- Approach to consolidate to one data point per site per sampling

- Average discussed. Simple decision when they agree well. What if they do not?
- Average by site and date
- Identify the replicates and compare
- If good agreement, use average

3. **Multiple locations per sampling date**
   - This was an MESL addition to the agenda
   - Objective is to best represent water quality at Station W2
   - Is there pseudo-replication by including numerous reference sites to represent W2 on a given date when an alternative (better?) approach might be to use the reference dataset to identify an surrogate W2 value for that date (or campaign if within a few days)
   - Option 1 → use all data as independent
   - Option 2 → consolidate by sampling date (or campaign) to get one estimate of "W2 background" for that date and then use those data for summary statistics
   - Jesse Sinclair ran scenarios on the spot (using R?)
     - D Copper Option 1 → N 184 95th 6.52 median 1.70
     - D Copper Option 2 → N 124 95th 4.94 median 1.46
     - Not much difference for TDS or NO3
   - Tim Barrett to apply this approach and indicate implications overall (e.g., lower BG for copper [but perhaps better representative of W2])
   - Alternative from Don MacDonald = only consider W2 in 2005?
   - Group … no, now that we have agreed on background dataset, stick with it
   - Check → in baseline ONLY (2005), how would Option 1 versus Option 2 compare to actual W2 (admittedly small dataset!)

4. **Dealing with Outliers**
   - Box plot approach suggested by Tim is appropriate.
   - MESL agrees with boxplot approach, indicates that there is more than one way, but that the 3IQR method is appropriate
   - Go ahead and complete the identification of potential outliers
   - Evaluate concurrent conditions
   - Provide list of outliers to Ryan Herbert at Minto Mine
   - Ryan can help decide what is feasible to investigate. If we can't investigate may have to make a (or practical to just toss)
   - Prepare a final list of proposed outliers to be eliminated
   - Provide list to SFN / MESL

5. **Method Detection Limits**
   - Significant issue for a number of COPCs
   - Pierre Stecko → issue could potentially be avoided by using non-parametric statistics
   - How does ½ substitution affect results if MDLs decrease in the future?
   - Pierre → acknowledge ABOVE in the document and propose a defensible way to deal with them. Part of our responsibility is to make sure that this set up in a way that can be easily

picked up in future reviews. Implications of method changes need to be identified and, to the extent possible, guidance provided on how they should be dealt with (e.g., Tim suggests the Kaplan-Meier method).

6. **Conditions / Binning**
   - Desirable not to bin
   - Evidence of seasonal patterns that indicate a need to "bin"
   - Investigated "under ice" versus "ice free"
   - Ice April to October = ice free; "under ice" = Nov 1 to Apr 1
   - Is under ice necessary? Frozen solid, no source?
   - AGREE … no need to apply under ice (background characterization of "under ice" period is not required
   - Ascending versus descending hydrograph (divisions from MESL 2013)
   - Box plots appear comparable for the analytes that we looked at (copper, nitrate, TDS)
   - Compare these two "bins" for each of the COPCs
   - Statistical contrast of the two groups
   - Discuss with SFN / MESL and make final decisions
   - It is possible that some analytes require binning whereas others do not

7. **Summary Statistics**
   - Two types – "Individual" (applies to grab sample) and "Central Tendency" (applies to average over a specified time frame [year?])
   - Individual
     - 95^{th} percentiles are appropriate
     - See attainment
     - Calculate following agreement on final dataset and approach (all previous items) with SFN / MESL
   - Central Tendency
     - Fairly brief discussion on this
     - PS indicated preference for non-parametric if feasible and defensible (e.g., 95^{th} percentile of medians), but key questions is whether data supports this in a defensible way
     - Calculate the 95^{th} percentile on medians; is this defensible with the given data
     - Calculate other measures of central tendency (UCL of the mean)
     - Compare the methods
     - Select more appropriate

8. **Attainment**
   - Individual (95^{th} percentiles)
     - Exceedances on 2 consecutive samples. If independent, 5% * 5% probability = 0.25%
     - Sufficient time between sample to have reasonable assurance
     - This approach allows a more rapid response than waiting to evaluate exceedance rates

    - Incorporate the final concept(s) into background report
- Central Tendency
    - Evaluate compliance annually. Does the annual median or mean fall within the range of background medians or means.
    - As above, must still recognize small chance of exceedance in the absence of any change.
    - Functionally, this could be an annual review item. Presumably, if the "central tendency" WQO has been exceeded, some "individual" exceedance has also occurred. The annual review should include investigation.
        - One analyte or many?
        - Frequency of exceedance of the "individual WQO"
        - Trending
        - Other?
    - Incorporate the final concept(s) into background report
- Bill Slater → these concepts work well when we are just considering future water quality relative to background, but become more complicated when considering a WQO that is a composite of background and the operation WQO.
- Don MacDonald – SFN would like to see WQO set somewhere between background and 50% of current operational WQO (WUL definition of assimilative capacity).

COPC List

- Don MacDonald – Would like to include TDS as a COPC with reasoning around making it easier to apply it is a communication piece with SFN. Don discussed having TDS as a major ion to follow and show the community that there are no problems with the WQ in Minto Creek.
- The group generally agrees that TDS is not expected to be an issue in closure as it relates to effects on aquatic life.
- No agreement on the inclusion of TDS in the COPC list was confirmed.
- Minto's team to discuss internally the merits of including TDS as a COPC.
- Don MacDonald – asked if we need to develop a framework to evaluating COPC's.
- Dylan MacGregor – Given the time remaining (until August 5th) it was probably not the best use of resources. The group as whole generally agreed.

Water Quality Prediction Appropriate Use

- Dylan MacGregor –Will have updated water quality predictions including CWTS performance expectations near end of month. Dylan has concerns regarding how water quality predictions will be used.
- Don MacDonald – we have to use predictions to determine where WQO should be set.
- Dylan MacGregor – we could use the current predictions as we have said we expect to have the CWTS performance to range from 0 to some upper percentage of removal therefore CWTS performance may not have an impact.
- Further discussions need to be had to agree on how best to use WQ predictions.

TIMING

- Items 1 to 6 by week of June 13th …
- Discuss week of June 20th ??

APPENDIX B

RAW WATER QUALITY DATA

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Total Coliforms	Fecal Coliforms	E.Coli	COD (mg/L)	Specific Conductivity in Field (µS/cm)	Radium - 226 (Bq/L)	CBOD (mg/L)	LC50 (%)	LT50 (h)	BOD5 (mg/L)	DO (mg/L)	Cond-F (µS/cm)	DO (%)	ORP (mV)
W1	5/27/2005														
W1	6/30/2005						<0.0050								
W1	7/28/2005						<0.0050								
W1	8/29/2005						<0.0050								
W1	9/28/2005						<0.0050								
W1	10/15/2005						<0.0050								
W2	5/27/2005														
W2	5/27/2005														
W2	6/30/2005						<0.0050								
W2	7/28/2005						<0.0050								
W2	8/30/2005						<0.0050								
W2	9/28/2005						0.007								
W2	10/15/2005						<0.0050								
W3	5/27/2005														
W3	6/30/2005						<0.0050								
W3	7/29/2005						<0.0050								
W3	8/29/2005						<0.0050								
W3	9/28/2005						<0.0050								
W3	10/15/2005						<0.0050								
W8	5/27/2005														
W8	6/30/2005														
W8	7/29/2005														
W8	8/30/2005														
W8	9/28/2005														
W8	10/15/2005														
W9	5/27/2005														
W9	6/30/2005						<0.0050								
W9	7/29/2005						0.007								
W9	8/29/2005						<0.0050								
W9	9/28/2005						<0.0050								
W9	10/15/2005						<0.0050								
W10	5/27/2005														
W10	5/27/2005														
W10	6/30/2005						<0.0050								
W10	7/29/2005						<0.0050								
W10	8/29/2005						<0.0050								
W10	9/28/2005						<0.0050								
W10	10/15/2005						0.007								
W6	5/27/2005														
W6	6/30/2005														
W6	7/29/2005														
W6	8/30/2005														
W6	9/28/2005														
W6	10/15/2005														
W6	6/2/2006														
W6	8/25/2006														
W6	9/6/2006														
W6	6/20/2007														
W6	9/18/2007														
W6	6/17/2008														
W6	9/18/2008														
W6	4/14/2009														
W6	4/30/2009												0.17	170	
W6	7/28/2009														
W6	8/6/2009														
W6	8/6/2009												4.25		
W6	8/18/2009														
W6	8/25/2009												415		
W6	10/7/2009														
W6	10/7/2009												485	108.5	
W6	10/25/2009												502	99.4	
W6	4/19/2010												283		
W6	5/16/2010												192.4		
W6	5/17/2010												124.4		
W6	5/19/2010												196.2		
W6	5/21/2010												164.3		
W6	5/23/2010												216		
W6	5/25/2010												177		
W6	5/29/2010												152.3		
W6	6/2/2010												448		
W6	6/4/2010												182.2		
W6	6/6/2010												191		
W6	6/14/2010												197		
W6	6/16/2010												287		
W6	6/18/2010												279		
W6	6/20/2010												279		
W6	6/22/2010												186	102	
W6	6/24/2010												187	101.2	
W6	6/26/2010														
W6	6/26/2010												189	99.4	
W6	6/28/2010												178	101.7	
W6	7/5/2010												164	101.4	
W6	8/5/2010												205	76.7	
W6	9/9/2010												205	79.6	
W6	10/8/2010												209	94.2	
W6	8/19/2011														
W6	7/17/2012					193.9						117	118.9	92.3	74.9
W6	7/30/2012					174.4						12.56	108.3	98.8	22.5
W6	8/9/2012					217.6						12.03	140.2	97.9	-22.9
W6	8/25/2012					196						11.94	120.1	92.8	134.6
W6	9/13/2012					171.5						13.7	95.8	94.3	64.9
W6	10/13/2012					194.5						13.83	101.6	94.7	27.8
W6	11/12/2012					365.5							190		249.6
W6	5/14/2013					53.6						13.22	28.1	92.1	137.8
W6	6/22/2013					172.6						12.33		88.6	79.4
W6	7/12/2013					180.5						11.38	109	87.7	93.3
W6	8/22/2013					204.1						11.81	127.8	102.4	22.8
W6	9/6/2013					201.8						11.85	122.7	98.7	20.4
W6	10/7/2013					191						14.61	104.1	102.3	36.2
W6	11/16/2013					196.2						8.31	102.6	68.3	98.6
W6	12/7/2013												231		
W7	5/27/2005														
W7	6/30/2005														
W7	7/29/2005														
W7	8/30/2005														
W7	9/28/2005														
W7	10/15/2005														
W7	6/2/2006														
W7	6/8/2006														

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Total Coliforms	Fecal Coliforms	E.Coli	COD (mg/L)	Specific Conductivity in Field (µS/cm)	Radium - 226 (Bq/L)	CBOD (mg/L)	LC50 (%)	LT50 (h)	BOD5 (mg/L)	DO (mg/L)	Cond-F (µS/cm)	DO (%)	ORP (mV)
W7	6/15/2006														
W7	6/15/2006														
W7	6/23/2006														
W7	6/28/2006														
W7	7/7/2006														
W7	7/12/2006														
W7	7/20/2006														
W7	7/26/2006														
W7	7/26/2006														
W7	8/2/2006														
W7	8/10/2006														
W7	8/25/2006														
W7	8/25/2006														
W7	8/30/2006														
W7	9/6/2006														
W7	9/13/2006														
W7	9/20/2006														
W7	9/28/2006														
W7	10/4/2006														
W7	10/12/2006														
W7	6/5/2007														
W7	6/5/2007														
W7	6/5/2007														
W7	6/20/2007														
W7	7/18/2007														
W7	7/18/2007														
W7	7/18/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	9/18/2007														
W7	10/30/2007														
W7	10/30/2007														
W7	10/30/2007														
W7	10/30/2007														
W7	4/22/2008														
W7	4/22/2008														
W7	4/22/2008														
W7	4/22/2008														
W7	6/3/2008														
W7	6/3/2008														
W7	6/3/2008														
W7	6/3/2008														
W7	6/17/2008														
W7	8/6/2008														
W7	8/6/2008														
W7	8/6/2008														
W7	8/6/2008														
W7	9/18/2008														
W7	10/28/2008														
W7	10/28/2008														
W7	10/28/2008														
W7	10/28/2008														
W7	4/30/2009						<0.006						37.6	183	
W7	6/8/2009												211		
W7	7/28/2009														
W7	8/6/2009												289		
W7	8/24/2009						0.01								
W7	8/25/2009												253		
W7	8/31/2009														
W7	10/7/2009														
W7	10/7/2009						<0.005						258	109	
W7	10/25/2009						<0.005						269	88	
W7	10/25/2009														
W7	4/19/2010												437		
W7	4/27/2010												67	59.2	
W7	4/29/2010												35	8.8	
W7	5/1/2010												82	102.2	
W7	5/3/2010												89	90.2	
W7	5/5/2010														
W7	5/7/2010												118	83.3	
W7	5/9/2010												171	129.2	
W7	5/10/2010												175	133.9	
W7	5/12/2010														
W7	5/12/2010														
W7	5/14/2010												145		
W7	5/16/2010												167.4		
W7	5/17/2010												161.6		
W7	5/19/2010												199.4		
W7	5/21/2010												191		
W7	5/23/2010												216		
W7	5/25/2010												224		
W7	5/27/2010												234		
W7	5/29/2010												243		
W7	5/31/2010												249		
W7	6/2/2010												254		
W7	6/4/2010												667		
W7	6/6/2010												262		
W7	6/14/2010												297		
W7	6/16/2010												364		
W7	6/18/2010												360		
W7	6/20/2010												342		
W7	6/22/2010												205	107.1	
W7	6/24/2010												211	105.6	
W7	6/26/2010												229	107.3	
W7	6/28/2010												245	105.1	
W7	7/5/2010												194	104.5	
W7	8/5/2010												240	78.1	
W7	9/12/2010														
W7	9/12/2010						<0.005						210	84.5	
W7	10/8/2010												235	100.3	
W7	4/27/2011														
W7	5/2/2011														
W7	6/1/2011														
W7	6/16/2011														
W7	7/1/2011														

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Total Coliforms	Fecal Coliforms	E.Coli	COD (mg/L)	Specific Conductivity in Field (µS/cm)	Radium - 226 (Bq/L)	CBOD (mg/L)	LC50 (%)	LT50 (h)	BOD5 (mg/L)	DO (mg/L)	Cond-F (µS/cm)	DO (%)	ORP (mV)
W7	8/2/2011														
W7	8/16/2011														
W7	8/19/2011														
W7	9/12/2011											18.55	204.1	133.5	
W7	10/6/2011											12.13	250.8	83.1	
W7	11/22/2011														
W7	1/24/2012														
W7	2/25/2012					296.9						10.92	155.2	74.6	
W7	3/5/2012					272.6						11.47	142.3	78.9	
W7	4/24/2012					75.6						15.55	39.5	106.3	
W7	5/6/2012														
W7	5/6/2012					88.8						8.53	46.4	58.4	116.8
W7	6/19/2012					202.6						12.45	121.6	95.1	44.5
W7	7/17/2012					244.2						12.31	152.9	97.4	45.6
W7	7/30/2012					235.1						12.64	142.6	97.6	16
W7	8/8/2012					268.5						12.65	164.5	98.5	-6.2
W7	8/23/2012					264.6						10.94	161.3	84.8	-2.4
W7	9/13/2012					224						13.04	126.2	94.1	76.7
W7	10/16/2012					258.9						17.1	137.8	118.3	18
W7	11/2/2012														
W7	12/29/2012					2107						9.38	1090	72.5	289.9
W7	4/30/2013					365.3							190.6	101.8	-50
W7	5/14/2013					59.7						13.46	31.1	92.4	109.7
W7	6/19/2013														
W7	6/19/2013												249		
W7	7/13/2013					250.7						11.92	148	90	108.7
W7	8/21/2013					280.6						12.06	168	101	32.7
W7	9/6/2013					277.3						11.37	166.3	93.5	38.1
W7	10/21/2013					286.3						11.57	149.3	77.7	16.6
W7	11/16/2013					197.6						7.17	104.3	61.8	94.7
W7	12/7/2013												198		
C4	5/6/2012					100.1						10.69	52.4	73.1	63.9
C4	8/1/2012					196.8						10.45	138.1	91.3	-38.1
C4	8/8/2012					246.3						11.52	165.3	97.1	-54.1
C4	8/25/2012					230.8						11.18	148.8	91	-21.8
C4	6/22/2013					223.8						10.58		87.9	93.7
C4	7/19/2013												260		
C4	8/22/2013					246.6						11.35	166.1	103.6	11.6
C4	9/6/2013					246.2						11.56	153.8	98.7	-28.8
C4	10/18/2013														
C4	10/18/2013					257.1						13.73	134.1	92.3	-20.1
C4	11/17/2013					274.1						4.27	142.7	28.4	101.5
C10	5/12/2012					144						13.64	82	99.9	45.3
C10	7/30/2012														
C10	7/30/2012					216.7						9.03	159.3	3	44.8
C10	8/10/2012					269						12.15	179	101.3	-28.1
C10	8/27/2012					257.4						11.33	170.7	94.1	27
C10	6/22/2013					242.7						9.91		91.1	85.9
C10	7/23/2013					227.6						11.31	153	95.3	97.7
C10	8/22/2013					304.4						11.25	206.6	101.4	143.1
C10	9/8/2013														
C10	9/8/2013					213.8						10.32	146.2	96.1	103.3
C10	10/18/2013					262.4						14.43	137.1	96.2	84.9

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Field pH (pH units)	Field Turbidity (NTU)	Field Temp (° C)	Sulphide (mg/L)	Dissolved SO4 (mg/L)	Total SO4 (mg/L)	TT CN (mg/L)	Chloride (mg/L)	Bromine (mg/L)	Fluoride (mg/L)	Chromium 6 (mg/L)	Chromium III (mg/L)	Cu-D-IH (mg/L)	Al-D-IH (mg/L)
W1	5/27/2005					19.7	12.8		<0.50						
W1	6/30/2005					22.4	22.4	0.011	1.2						
W1	7/28/2005						19.1	0.0054	0.64						
W1	8/29/2005					12.4	12.4	0.0065	0.52						
W1	9/28/2005					11.8	11.8	0.0118	0.73						
W1	10/15/2005						13.9	0.125	0.91						
W2	5/27/2005						12.8		<0.50						
W2	5/27/2005					9.64			<0.5						
W2	6/30/2005					22.1	22.1	<0.0050	1.23						
W2	7/28/2005						19	<0.0050	0.66						
W2	8/30/2005					12.6	12.6	0.0053	0.53						
W2	9/28/2005					11.6	11.6	0.0106	0.72						
W2	10/15/2005						14	0.0101	0.88						
W3	5/27/2005					2.4	19.7		<0.50						
W3	6/30/2005					45.5	45.5	<0.0050	0.86						
W3	7/29/2005						29.6	<0.0050	<0.50						
W3	8/29/2005					25.5	25.5	0.0091	<0.50						
W3	9/28/2005					23.9	23.9	0.013	<0.50						
W3	10/15/2005						24.7	0.014	<0.50						
W8	5/27/2005					166	7.62		<0.50						
W8	6/30/2005					<1	<1.0		0.86						
W8	7/29/2005						7.08		<0.50						
W8	8/30/2005					5.1	5.1		<0.50						
W8	9/28/2005					7.72	7.72		<0.50						
W8	10/15/2005						24.8		<0.50						
W9	5/27/2005					35.3	11.8		<0.50						
W9	6/30/2005					18.4	18.4	<0.0050	<5.0						
W9	7/29/2005						16.5	<0.0050	<0.50						
W9	8/29/2005					17.9	17.9	0.0124	<0.50						
W9	9/28/2005					16.7	16.7	0.0128	<0.50						
W9	10/15/2005						17.4	0.0164	<0.50						
W10	5/27/2005						<0.50		<0.50						
W10	5/27/2005								<0.5						
W10	6/30/2005					<10	<10	<0.0050	<5.0						
W10	7/29/2005						1.56	<0.0050	<0.50						
W10	8/29/2005					1.82	1.82	0.0226	<0.50						
W10	9/28/2005					4.13	4.13	0.0136	<0.50						
W10	10/15/2005						4.19	0.018	<0.50						
W6	5/27/2005					11.8	2.4		<0.50						
W6	6/30/2005					<1	<1.0		0.73						
W6	7/29/2005						29.2		<0.50						
W6	8/30/2005					2.62	2.62		<0.50						
W6	9/28/2005					4.04	4.04		0.5						
W6	10/15/2005						4.69		0.52						
W6	6/2/2006					1.88			<0.5	<0.05	0.176				
W6	8/25/2006					2.4									
W6	9/6/2006					2.3									
W6	6/20/2007					2.1			0.4						
W6	9/18/2007					2.51			0.59						
W6	6/17/2008					1.64			0.29						
W6	9/18/2008					3.29			0.41						
W6	4/14/2009					6.28			6.32						
W6	4/30/2009	8.2		5.6		7.44			0.76						
W6	7/28/2009					42.9			6.17						
W6	8/6/2009														
W6	8/6/2009	7.25		9.6		48.4			9.84						
W6	8/18/2009														
W6	8/25/2009	7.82		10.8											
W6	10/7/2009					62.2			27.8						
W6	10/7/2009	8.14		4.22		63.3			33.3						
W6	10/25/2009	8.12		2.36											
W6	4/19/2010	6.6		1		30.3			<0.02						
W6	5/16/2010	7.52		1.1		6.2			1	<0.4	0.56				
W6	5/17/2010	7.7		1.1		8.1			0.8	<0.4	0.22				
W6	5/19/2010	7.87		1.1		13			1	<0.4	0.27				
W6	5/21/2010	8.01		2.6		7.2			0.9	<0.4	0.26				
W6	5/23/2010	7.34		1		16			2.1	<0.4	0.31				
W6	5/25/2010	7.48		1		17			1.4	<0.4	0.29				
W6	5/29/2010	7.79		1.4		3.5			0.8	<0.4	0.24				
W6	6/2/2010	7.38		1.5		49			6.4	<0.4	0.43				
W6	6/4/2010	7.87		2.6		<0.5			<0.5	<0.4	0.2				
W6	6/6/2010			1.6		<0.5			<0.5	<0.4	0.21				
W6	6/14/2010	6.65		3.8		1.4			<0.5	<0.4	0.23				
W6	6/16/2010	6.8		2.5		<0.5			<0.5	<0.4	0.22				
W6	6/18/2010	7.03		2.9		<0.5			<0.5	<0.4	0.23				
W6	6/20/2010	6.38		3.1		0.7			<0.5	<0.4	0.23				
W6	6/22/2010	7.8	-0.2	1.92		<0.5			<0.5	<0.4	0.23				
W6	6/24/2010	7.91	-1.2	2.38		<0.5			<0.5	<0.4	0.25				
W6	6/26/2010					<0.5			<0.5	<0.4	0.24				
W6	6/26/2010	7.78	-1	2.45											
W6	6/28/2010	7.99	5.2	2.79		<0.5			<0.5	<0.4	0.22				
W6	7/5/2010	7.99	23.1	3.71		<0.5			<0.5	<0.4	0.2				
W6	8/5/2010	7.96	-0.4	6.72		<0.5			0.5	<0.4	0.23				
W6	9/9/2010	7.99	9.5	5.1		<0.5			0.7	<0.4	0.26				
W6	10/8/2010	7.75	-1.3	-0.18		2.1			<0.5	<0.4	0.21				
W6	8/19/2011														
W6	7/17/2012	7.92		4.8		1.82			1		0.25				
W6	7/30/2012	7.89		5.2		<0.50			0.58		0.3				
W6	8/9/2012	8.09		6.4		0.64			1.1		0.28				
W6	8/25/2012	7.76		4.7		<0.50			0.92		0.28				
W6	9/13/2012	7.95		1.9		0.62			1.1		0.22				
W6	10/13/2012	8.3		0		1.52			0.87		0.23				
W6	11/12/2012	6.69		-0.1		5.81			1.1		0.23				
W6	5/14/2013	7.6		0.1		<0.50			1.5		0.083				
W6	6/22/2013	7.61		1.6		1.88			1.1		0.22				
W6	7/12/2013	7.69		4.4		1.27			1		0.23				
W6	8/22/2013	7.81		5.4		0.53			0.84		0.23				
W6	9/6/2013	7.60		4.5		0.62			1.2		0.26				
W6	10/7/2013	7.66		1.2		1.89			0.8		0.25				
W6	11/16/2013	7.96		-0.1		1.48			0.78		0.24				
W6	12/7/2013	7.50		0		5.36			1.4		0.28				
W7	5/27/2005					<0.5	6.18		<0.50						
W7	6/30/2005					10.2	10.2		0.92						
W7	7/29/2005						29.9		<0.50						
W7	8/30/2005					6.02	6.02		<0.50						
W7	9/28/2005					5.62	5.62		<0.50						
W7	10/15/2005						7.47		<0.50						
W7	6/2/2006					2.98			<0.5	<0.05	0.155				
W7	6/8/2006														

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Field pH (pH units)	Field Turbidity (NTU)	Field Temp (° C)	Sulphide (mg/L)	Dissolved SO4 (mg/L)	Total SO4 (mg/L)	TT CN (mg/L)	Chloride (mg/L)	Bromine (mg/L)	Fluoride (mg/L)	Chromium 6 (mg/L)	Chromium III (mg/L)	Cu-D-IH (mg/L)	Al-D-IH (mg/L)
W7	6/15/2006														
W7	6/15/2006					10.5									
W7	6/23/2006					15.1									
W7	6/28/2006														
W7	7/7/2006					10.8									
W7	7/12/2006					10.4									
W7	7/20/2006					10.5									
W7	7/26/2006														
W7	7/26/2006					11.6									
W7	8/2/2006					10									
W7	8/10/2006					10									
W7	8/25/2006														
W7	8/25/2006					11									
W7	8/30/2006					10.5									
W7	9/6/2006					10.3									
W7	9/13/2006					10									
W7	9/20/2006					11									
W7	9/28/2006					11									
W7	10/4/2006					10.9									
W7	10/12/2006					11.3									
W7	6/5/2007					11			0.7						
W7	6/5/2007					11			0.7						
W7	6/5/2007					11			0.7						
W7	6/20/2007					12.2			0.6						
W7	7/18/2007					11.9			0.6						
W7	7/18/2007					11.9			0.6						
W7	7/18/2007					11.9			0.6						
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007					10.3			0.3						
W7	8/24/2007					10.2			0.3						
W7	8/24/2007					10.3			0.3						
W7	9/18/2007					8.21			0.46						
W7	10/30/2007					13			0.96						
W7	10/30/2007					13.6			1.32						
W7	10/30/2007					13			0.88						
W7	10/30/2007					13			0.91						
W7	4/22/2008					27.6			0.8						
W7	4/22/2008					27.8			0.62						
W7	4/22/2008					27.8			0.78						
W7	4/22/2008					27.8			0.62						
W7	6/3/2008					10.4			0.38						
W7	6/3/2008					10.4			0.37						
W7	6/3/2008					10.4			0.38						
W7	6/3/2008														
W7	6/17/2008					12.9			0.42						
W7	8/6/2008					9.89			0.29						
W7	8/6/2008					9.98			0.29						
W7	8/6/2008					9.98			0.3						
W7	8/6/2008														
W7	9/18/2008					5.44			0.57						
W7	10/28/2008					9.43			0.34						
W7	10/28/2008					9.36			0.31						
W7	10/28/2008					9.44			0.31						
W7	10/28/2008														
W7	4/30/2009	7.6		0.4		0.65			0.38						
W7	6/8/2009	7.74		3.27		8.6			4.1						
W7	7/28/2009					13			0.82						
W7	8/6/2009	7.5		4.4		13			0.39						
W7	8/24/2009														
W7	8/25/2009	8.04		3.02											
W7	8/31/2009														
W7	10/7/2009														
W7	10/7/2009	8.24		-0.8		9.9			0.61						
W7	10/25/2009	8.23		0.2											
W7	10/25/2009														
W7	4/19/2010	7.31		1		8.2			0.91						
W7	4/27/2010	7.65	0	-0.07		7			1.4	<0.4	0.08				
W7	4/29/2010	7.72	-0.4	-0.07		<5			0.9	<0.4	0.09				
W7	5/1/2010	7.15	6.9	-0.06		<5			1.2	<0.4	0.11				
W7	5/3/2010	7.74	6.3	0.04		<0.5			0.8	<0.4	0.11				
W7	5/5/2010					<0.5			0.9	<0.4	0.13				
W7	5/7/2010	7.95	5.6	0.03		<0.5			0.8	<0.4	0.13				
W7	5/9/2010	8.03	-0.8	0.8		9.1			1.2	<0.4	0.22				
W7	5/10/2010	8.06	-0.9	0.07		3.9			0.6	<0.4	0.2				
W7	5/12/2010														
W7	5/12/2010					5.4			<0.5	<0.4	0.11				
W7	5/14/2010	7.26		1.1		5.2			<0.5	<0.4	0.18				
W7	5/16/2010	7.32		1.10		7.6			0.6	<0.4	0.2				
W7	5/17/2010	7.52		1		5.6			<0.5	<0.4	0.19				
W7	5/19/2010	7.62		1		9.7			<0.5	<0.4	0.25				
W7	5/21/2010	8.02		1.2		9.3			0.8	<0.4	0.26				
W7	5/23/2010	7.61		1		9.9			1	<0.4	0.28				
W7	5/25/2010	7.24		1.2		11			0.8	<0.4	0.28				
W7	5/27/2010	7.53		3		12			<0.5	<0.4	0.3				
W7	5/29/2010	7.73		1		12			0.7	<0.4	0.32				
W7	5/31/2010	7.83		1.2		12			0.7	<0.4	0.32				
W7	6/2/2010	7.26		1.4		12			<0.5	<0.4	0.31				
W7	6/4/2010	7.48		3.1		13			0.8	<0.4	0.31				
W7	6/6/2010			1.9		12			<0.5	<0.4	0.32				
W7	6/14/2010			3.8		13			<0.5	<0.4	0.35				
W7	6/16/2010	6.78		2.6		14			<0.5	<0.4	0.35				
W7	6/18/2010	7.03		3		12			<0.5	<0.4	0.34				
W7	6/20/2010	7.1		3.4		10			0.5	<0.4	0.31				
W7	6/22/2010	7.92	135	3.25		8.2			<0.5	<0.4	0.25				
W7	6/24/2010	7.95	30	3.55		6.4			<0.5	<0.4	0.28				
W7	6/26/2010	8.03	9.7	3.12		7.8			<0.5	<0.4	0.3				
W7	6/28/2010	7.89	2.3	2.94		9.7			<0.5	<0.4	0.29				
W7	7/5/2010	8.02	35.4	4.48		2.5			<0.5	<0.4	0.22				
W7	8/5/2010	7.91	3.1	5.57											
W7	9/12/2010														
W7	9/12/2010	8.02	0.8	4.06		6.4			0.7	<0.4	0.22				
W7	10/8/2010	7.85	-2.7	-0.18		8.8			<0.5	<0.4	0.23				
W7	4/27/2011					<5			1.9		0.1				
W7	5/2/2011					<5			1.7		0.08				
W7	6/1/2011					6.1			0.5		0.24				
W7	6/16/2011														
W7	7/1/2011					<0.5									

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Field pH (pH units)	Field Turbidity (NTU)	Field Temp (° C)	Sulphide (mg/L)	Dissolved SO4 (mg/L)	Total SO4 (mg/L)	TT CN (mg/L)	Chloride (mg/L)	Bromine (mg/L)	Fluoride (mg/L)	Chromium 6 (mg/L)	Chromium III (mg/L)	Cu-D-IH (mg/L)	Al-D-IH (mg/L)
W7	8/2/2011								1.8		0.14				
W7	8/16/2011														
W7	8/19/2011														
W7	9/12/2011	7.92		1.9					1.3						
W7	10/6/2011	7.65		0		8.9			1						
W7	11/22/2011								0.8						
W7	1/24/2012					18			0.8		0.43				
W7	2/25/2012	7.16		0.00		17.8			0.9		0.42				
W7	3/5/2012	6.72		0		19			0.5		0.42				
W7	4/24/2012	7.43		0		<0.50			1.4		0.1				
W7	5/6/2012					<0.50			0.58		0.12				
W7	5/6/2012	7.1		0		<0.50			<0.50		0.12				
W7	6/19/2012	7.59		4.1		<0.50			1.1		0.17				
W7	7/17/2012	7.76		5.4		4.63			0.96		0.23				
W7	7/30/2012	7.86		4.4		5.31			0.58		0.28				
W7	8/8/2012	8.44		4.7		5.43			1.2		0.29				
W7	8/23/2012	7.79		4.6		5.23			1.2		0.27				
W7	9/13/2012	7.98		1.9		3.85			1.2		0.18				
W7	10/16/2012	7.89		0.4		7.47			1.2		0.19				
W7	11/2/2012					14.4			0.82		0.29				
W7	12/29/2012	7.97		-0.3		29.1			1.9		0.43				
W7	4/30/2013	8.18		0		24.1			1.4		0.53				
W7	5/14/2013	8.2		0		<0.50			1.5		0.064				
W7	6/19/2013					8.07			1.2		0.25				
W7	6/19/2013	8.6		4.6		7.87			1.1		0.25				
W7	7/13/2013	7.74		3.6		9.73			0.99		0.26				
W7	8/21/2013	7.81		4.2		8.32			0.93		0.27				
W7	9/6/2013	7.75		4		5.66			1.3		0.24				
W7	10/21/2013	7.58		0		10.3			1.3		0.25				
W7	11/16/2013	7.69		-0.1		15.1			1		0.35				
W7	12/7/2013	6.30		0		16.7			1.2		0.31				
C4	5/6/2012	7.51		0		<0.50			0.54		0.14				
C4	8/1/2012	7.75		9.3		3.37			0.89		0.2				
C4	8/8/2012	7.81		7.8		<5.0			<5.0		0.22				
C4	8/25/2012	7.81		6.3		2.19			1.4		0.22				
C4	6/22/2013	7.68		7.3		10.6			1.4		0.27				
C4	7/19/2013	8.20		9.6		7.45			1.5		0.27				
C4	8/22/2013	7.88		7.7		4.75			1.5		0.21				
C4	9/6/2013	7.69		5.3		4.27			1.3		0.19				
C4	10/18/2013					11.1			1.2		0.19				
C4	10/18/2013	7.16		-0.1		38.1			2.4		0.2				
C4	11/17/2013	7.17		-0.1		6.33			2.4		0.29				
C10	5/12/2012	7.89		2.4		<0.50			1.1		0.2				
C10	7/30/2012					<0.50			1.2		0.29				
C10	7/30/2012	8.16		11.2											
C10	8/10/2012	8.71		7.4		<0.50			0.75		4				
C10	8/27/2012	8.3		7.4		<0.50			2		0.24				
C10	6/22/2013	8.29		11.7		0.97			1.3		0.21				
C10	7/23/2013	8.31		7.8		<0.50			1.3		0.21				
C10	8/22/2013	8.18		8.2		<0.50			1.6		0.21				
C10	9/8/2013					<0.50			1.6		0.2				
C10	9/8/2013	8.16		9		<0.50			1.3		0.2				
C10	10/18/2013	7.78		0		3.24			1.5		0.17				

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Cd-D-IH (mg/L)	Se-D-IH (mg/L)	Ag-D (mg/L)	Dissolved Al (mg/L)	Dissolved As (mg/L)	Dissolved Ba (mg/L)	Dissolved B (mg/L)	Dissolved Be (mg/L)	Dissolved Bi (mg/L)	Dissolved Ca (mg/L)	Dissolved Cd (mg/L)	Dissolved Co (mg/L)	Dissolved Cr (mg/L)	Dissolved Cu (mg/L)
W1	5/27/2005			<0.000020	0.0176	0.00051	0.049	<0.10	<0.0010		25	<0.000050	<0.00030	<0.0010	0.0034
W1	6/30/2005			<0.000010	0.0065	0.00048	0.0554	<0.010	<0.00050	<0.00050	37.5	<0.000050	<0.00010	<0.00050	0.00204
W1	7/28/2005			<0.000010	0.0072	0.00048	0.0575	<0.010	<0.00050	<0.00050	40.4	<0.000050	<0.00010	<0.00050	0.00211
W1	8/29/2005			<0.000020	0.0106	0.0006	0.058	<0.10	<0.0010		35.9	<0.000017	<0.00030	<0.0010	0.0024
W1	9/28/2005			<0.000020	0.0143	0.00058	0.055	<0.10	<0.0010		35.3	<0.000017	<0.00030	<0.0010	0.0026
W1	10/15/2005			<0.000010	0.0137	0.00054	0.0492	<0.010	<0.00050	<0.00050	36.8	<0.000050	<0.00010	<0.00050	0.00279
W2	5/27/2005			<0.000020	0.0163	<0.00050	0.05	<0.10	<0.0010		25.6	<0.000050	<0.00030	<0.0010	0.0033
W2	5/27/2005				0.0108		0.032				20.3				
W2	6/30/2005			<0.000010	0.0073	0.00044	0.0574	<0.010	<0.00050	<0.00050	36.9	<0.000050	<0.00010	<0.00050	0.0021
W2	7/28/2005			<0.000010	0.0071	0.00044	0.0604	<0.010	<0.00050	<0.00050	39.6	<0.000050	<0.00010	<0.00050	0.00208
W2	8/30/2005			<0.000020	0.0119	0.00058	0.059	<0.10	<0.0010		35.7	<0.000017	<0.00030	<0.0010	0.0024
W2	9/28/2005			<0.000020	0.0141	0.00059	0.055	<0.10	<0.0010		35.2	<0.000017	<0.00030	<0.0010	0.0025
W2	10/15/2005			<0.000010	0.0129	0.00053	0.0492	<0.010	<0.00050	<0.00050	35.9	<0.000050	<0.00010	<0.00050	0.00271
W3	5/27/2005			<0.000020	0.0267	<0.00050	0.041	<0.10	<0.0010		22	<0.000050	<0.00030	<0.0010	0.0066
W3	6/30/2005			<0.000010	0.008	0.00034	0.0507	<0.010	<0.00050	<0.00050	38.5	<0.000050	<0.00010	<0.00050	0.00336
W3	7/29/2005			<0.000010	0.0113	0.00041	0.0553	<0.010	<0.00050	<0.00050	38	<0.000050	0.0001	<0.00050	0.00478
W3	8/29/2005			<0.000020	0.0184	<0.00050	0.057	<0.10	<0.0010		31.8	<0.000017	<0.00030	<0.0010	0.005
W3	9/28/2005			<0.000020	0.0257	<0.00050	0.055	<0.10	<0.0010		30.4	<0.000017	<0.00030	<0.0010	0.0057
W3	10/15/2005			<0.000010	0.0223	0.00043	0.0475	<0.010	<0.00050	<0.00050	31.6	<0.000050	0.00016	<0.00050	0.00608
W8	5/27/2005			<0.000020	0.0427	<0.00050	0.039	<0.10	<0.0010		16.6	<0.000050	<0.00030	<0.0010	0.0141
W8	6/30/2005			<0.000010	0.0193	0.00043	0.0466	<0.010	<0.00050	<0.00050	24.8	<0.000050	<0.00010	0.00061	0.00942
W8	7/29/2005			<0.000010	0.0201	0.00045	0.0538	<0.010	<0.00050	<0.00050	30.9	<0.000050	0.00014	0.00052	0.00863
W8	8/30/2005			<0.000020	0.0259	<0.00050	0.054	<0.10	<0.0010		26.3	<0.000017	<0.00030	<0.0010	0.0081
W8	9/28/2005			<0.000020	0.0305	<0.00050	0.049	<0.10	<0.0010		25.3	<0.000017	<0.00030	<0.0010	0.0081
W8	10/15/2005			0.000031	0.0258	0.0005	0.0479	<0.010	<0.00050	<0.00050	26.8	<0.000050	0.00028	<0.00050	0.00561
W9	5/27/2005			<0.000020	0.0718	<0.00050	0.042	<0.10	<0.0010		12.9	<0.000050	<0.00030	<0.0010	0.0035
W9	6/30/2005			<0.000010	0.0345	0.00054	0.0525	<0.010	<0.00050	<0.00050	18.9	<0.000050	0.00018	<0.00050	0.00184
W9	7/29/2005			<0.000010	0.0298	0.00057	0.056	<0.010	<0.00050	<0.00050	25.9	<0.000050	0.00018	<0.00050	0.00334
W9	8/29/2005			<0.000020	0.0449	0.00058	0.055	<0.10	<0.0010		19.1	0.000022	<0.00030	<0.0010	0.0017
W9	9/28/2005			<0.000020	0.0578	<0.00050	0.049	<0.10	<0.0010		17.7	<0.000017	<0.00030	<0.0010	0.0022
W9	10/15/2005			<0.000010	0.056	0.00039	0.0421	<0.010	<0.00050	<0.00050	16.6	<0.000050	0.00016	<0.00050	0.00209
W10	5/27/2005			<0.000020	0.135	<0.00050	0.052	<0.10	<0.0010		13.1	<0.000050	<0.00030	<0.0010	0.0184
W10	5/27/2005			<0.00002		<0.0005		<0.1	<0.001			<0.00005	<0.0003	<0.001	<0.001
W10	6/30/2005			<0.000010	0.113	0.00059	0.0396	<0.010	<0.00050	<0.00050	13.1	<0.000050	0.00046	0.00062	0.00472
W10	7/29/2005			<0.000010	0.0683	0.00037	0.0369	<0.010	<0.00050	<0.00050	14.7	<0.000050	0.00011	<0.00050	0.00437
W10	8/29/2005			<0.000020	0.0433	<0.00050	0.044	<0.10	<0.0010		18.2	<0.000017	<0.00030	<0.0010	0.004
W10	9/28/2005			<0.000020	0.0302	<0.00050	0.057	<0.10	<0.0010		25.8	<0.000017	<0.00030	<0.0010	0.0033
W10	10/15/2005			<0.000010	0.0361	0.0003	0.0487	<0.010	<0.00050	<0.00050	25.4	<0.000050	<0.00010	<0.00050	0.00412
W6	5/27/2005			<0.000020	0.0144	<0.00050	0.035	<0.10	<0.0010		17.9	<0.000050	<0.00030	<0.0010	0.0016
W6	6/30/2005			<0.000010	0.0056	0.00039	0.0429	<0.010	<0.00050	<0.00050	25.2	<0.000050	<0.00010	<0.00050	0.00101
W6	7/29/2005			<0.000010	0.007	0.00026	0.0668	<0.010	<0.00050	<0.00050	46.9	<0.000050	<0.00010	<0.00050	0.00247
W6	8/30/2005			<0.000020	0.0067	<0.00050	0.053	<0.10	<0.0010		29	<0.000017	<0.00030	<0.0010	0.001
W6	9/28/2005			<0.000020	0.0067	<0.00050	0.048	<0.10	<0.0010		28.8	<0.000017	<0.00030	<0.0010	0.0011
W6	10/15/2005			<0.000010	0.0769	0.00042	0.0419	<0.010	<0.00050	<0.00050	26.7	<0.000050	<0.00010	<0.00050	0.00108
W6	6/2/2006			<0.00001	0.0252	0.00044	0.0315	<0.01	<0.0005	<0.0005	17.8	<0.00005	<0.0001	<0.0005	0.00175
W6	8/25/2006										31.7				
W6	9/6/2006			<0.0001	0.008	0.0005	0.051	<0.002	<0.0001	<0.0005	30	<0.00001	0.0001	0.0006	0.001
W6	6/20/2007														
W6	9/18/2007														
W6	6/17/2008														
W6	9/18/2008			<0.00001	0.012	0.0004	0.039	<0.004	<0.00004		23.7	<0.00001	0.00022	0.0013	0.002
W6	4/14/2009			<0.00001	0.038	0.0004	0.046	0.017	<0.00004	<0.0001	23.6	0.00002	0.00033	0.0004	0.043
W6	4/30/2009			<0.00001	0.097	0.0003	0.026	0.007	<0.00004	<0.0001	14.5	0.00023	0.00014	0.0007	0.008
W6	7/28/2009														
W6	8/6/2009														
W6	8/6/2009														
W6	8/18/2009			<0.00001	0.03	0.0004	0.101	0.023	<0.00004	<0.001	46.1	<0.00001	0.00014	0.001	0.004
W6	8/25/2009														
W6	10/7/2009														
W6	10/7/2009														
W6	10/25/2009			<0.00002	0.023	0.0003	0.118	<0.05	<0.0001	<0.001	57.9	0.00001	<0.0005	<0.001	0.0021
W6	4/19/2010			<0.00001	0.024	0.0002	0.048	0.009	<0.00004	<0.001	32.3	0.00003	0.00016	<0.0004	0.019
W6	5/16/2010			<0.00002	<0.003	0.0004	0.025	<0.05	<0.0001	<0.001	25.4	<0.00001	<0.0005	<0.001	0.0015
W6	5/17/2010			<0.00002	0.009	0.0003	0.033	<0.05	<0.0001	<0.001	20.8	<0.00001	<0.0005	<0.001	0.0022
W6	5/19/2010			<0.00002	0.004	0.0003	0.039	<0.05	<0.0001	<0.001	26	<0.00001	<0.0005	<0.001	0.0027
W6	5/21/2010			<0.0001	<0.01	<0.0004	0.04	<0.1	<0.0002	<0.001	25	<0.0001	<0.0005	<0.002	0.003
W6	5/23/2010			<0.0001	<0.01	<0.0004	0.046	<0.1	<0.0002	<0.001	28	<0.0001	<0.0005	<0.002	0.003
W6	5/25/2010			<0.0001	<0.01	<0.001	0.045	<0.1	<0.0005	<0.001	29	<0.0001	<0.0005	<0.002	0.006
W6	5/29/2010			<0.0001	<0.01	<0.0004	0.037	<0.1	<0.0002	<0.001	21	<0.0001	<0.0005	<0.002	0.003
W6	6/2/2010			<0.0001	<0.01	<0.0004	0.074	<0.1	<0.0002	<0.001	56	<0.0001	<0.0005	<0.002	0.001
W6	6/4/2010			<0.0001	<0.01	<0.0004	0.043	<0.1	<0.0002	<0.001	23	<0.0001	<0.0005	<0.002	<0.001
W6	6/6/2010			<0.0001	<0.01	0.0004	0.046	<0.1	<0.0002	<0.001	27	<0.0001	<0.0005	<0.002	0.001
W6	6/14/2010			<0.0001	<0.01	<0.0004	0.038	<0.1	<0.0002	<0.001	25	<0.0001	<0.0005	<0.002	0.001
W6	6/16/2010			<0.0001	<0.01	0.0005	0.044	<0.1	<0.0002	<0.001	25	<0.0001	<0.0005	<0.002	0.006
W6	6/18/2010			<0.0001	<0.01	<0.0004	0.044	<0.1	<0.0002	<0.001	26	<0.0001	<0.0005	<0.002	0.001
W6	6/20/2010			<0.0001	<0.01	<0.0004	0.043	<0.1	<0.0002	<0.001	25	<0.0001	<0.0005	<0.002	<0.001
W6	6/22/2010			<0.0001	<0.01	<0.0004	0.044	<0.1	<0.0002	<0.001	25	<0.0001	<0.0005	<0.002	<0.001
W6	6/24/2010			<0.0001	<0.01	<0.0004	0.043	<0.1	<0.0002	<0.001	26	<0.0001	<0.0005	<0.002	<0.001
W6	6/26/2010			<0.0001	<0.01	<0.0004	0.045	<0.1	<0.0002	<0.001	25	<0.0001	<0.0005	<0.002	<0.001
W6	6/26/2010														
W6	6/28/2010			<0.0001	<0.01	<0.0004	0.044	<0.1	<0.0002	<0.001	25	<0.0001	<0.0005	<0.002	<0.001
W6	7/5/2010			<0.0001	0.015	0.0004	0.047	<0.1	<0.0002	<0.001	22	<0.0001	<0.0005	<0.002	0.001
W6	8/5/2010			<0.0001	0.012	0.0005	0.056	<0.1	<0.0002	<0.001	31	<0.0001	<0.0005	<0.002	0.002
W6	9/9/2010			<0.0001	0.011	0.0004	0.053	<0.1	<0.0002	<0.001	30	<0.0001	<0.0005	<0.002	<0.001
W6	10/8/2010			<0.00002	0.008	0.0004	0.05	<0.05	<0.0001	<0.001	28.8	<0.00001	<0.0005	<0.001	0.0008
W6	8/19/2011			<0.00002	0.01	0.0003	0.077	<0.05	<0.0001	<0.001	53.6	<0.00001	<0.0005	<0.001	0.0025
W6	7/17/2012			<0.000020	0.0161	0.00053	0.0597	<0.05	<0.00010	<0.0010	29.7	<0.000010	<0.00050	<0.0010	0.00105
W6	7/30/2012			<0.000020	0.0186	0.00059	0.0601	<0.05	<0.00010	<0.0010	28.9	<0.000010	<0.00050	<0.0010	0.00089
W6	8/9/2012			<0.000020	0.0173	0.00081	0.0681	<0.05	<0.00010	<0.0010	31.9	<0.000010	<0.00050	<0.0010	0.00083
W6	8/25/2012			<0.000020	0.0138	0.00081	0.063	<0.05	<0.00010	<0.0010	28.9	<0.000010	<0.00050	<0.0010	0.00103
W6	9/13/2012			<0.000020	0.0169	0.00047	0.0451	<0.05	<0.00010	<0.0010	24.1	<0.000010	<0.00050	<0.0010	0.00145
W6	10/13/2012														
W6	11/12/2012			<0.000020	0.0062	0.00039	0.0697	<0.05	<0.00010	<0.0010	41.6	<0.000010	<0.00050	<0.0010	0.00143
W6	5/14/2013			<0.000020	0.0591	0.00037	0.0195	<0.05	<0.00010	<0.0010	8.75	0.000012	<0.00050	<0.0010	0.0035
W6	6/22/2013			<0.000020	0.0095	0.00053	0.0493	<0.05	<0.00010	<0.0010	26.4	<0.000010	<0.00050	<0.0010	0.00096
W6	7/12/2013			<0.000020	0.0066	0.00046	0.0438	<0.05	<0.00010	<0.0010	24.1	<0.000010	<0.00050	<0.0010	0.00095
W6	8/22/2013			<0.000020	0.0084	0.00052	0.0495	<0.05	<0.00010	<0.0010	28.3	<0.000010	<0.00050	<0.0010	0.00119
W6	9/6/2013			<0.000020	0.0083	0.00056	0.0499	<0.05	<0.00010	<0.0010	28.4	<0.000010	<0.00050	<0.0010	0.00089
W6	10/7/2013			<0.000020	0.0132	0.00051	0.0518	<0.05	<0.00010	<0.0010	30.7	<0.000010	<0.00050	<0.0010	0.00097
W6	11/16/2013			<0.000020	0.0057	0.00048	0.0582	<0.05	<0.00010	<0.0010	33.2	<0.000010	<0.00050	<0.0010	0.00084
W6	12/7/2013			<0.000020	0.0041	0.00061	0.0725	<0.05	<0.00010	<0.0010	44.6	0.000011	<0.00050	<0.0010	0.00156
W7	5/27/2005			<0.000020	0.0236	<0.00050	0.045	<0.10	<0.0010		19.5	<0.000050	<0.00030	<0.0010	0.0018
W7	6/30/2005			<0.000010	0.0086	0.00042	0.0609	<0.010	<0.00050	<0.00050	30	<0.000050	<0.00010	<0.00050	0.00119
W7	7/29/2005			<0.000010	0.0096	0.00035	0.061	<0.010	<0.00050	<0.00050	41.3	<0.000050	<0.00010	<0.00050	0.00404
W7	8/30/2005			<0.000020	0.0156	0.00051	0.062	<0.10	<0.0010		27.8	<0.000017	<0.00030	<0.0010	0.0014
W7	9/28/2005			<0.000020	0.0186	0.00054	0.055	<0.10	<0.0010		27.4	<0.000017	<0.00030	<0.0010	0.0016
W7	10/15/2005			<0.000010	0.0155	0.00048	0.0502	<0.010	<0.00050	<0.00050	27.1	<0.000050	0.00011	<0.00050	0.00127
W7	6/2/2006			<0.00001	0.0714	0.00062	0.0341	<0.01	<0.0005	<0.0005	14.4	<0.00005	0.00033	0.00055	0.00212
W7	6/8/2006														

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Cd-D-IH (mg/L)	Se-D-IH (mg/L)	Ag-D (mg/L)	Dissolved Al (mg/L)	Dissolved As (mg/L)	Dissolved Ba (mg/L)	Dissolved B (mg/L)	Dissolved Be (mg/L)	Dissolved Bi (mg/L)	Dissolved Ca (mg/L)	Dissolved Cd (mg/L)	Dissolved Co (mg/L)	Dissolved Cr (mg/L)	Dissolved Cu (mg/L)
W7	6/15/2006														
W7	6/15/2006			<0.0001	0.013	0.0004	0.063	0.003	<0.0001	<0.0005	26.9	<0.00001	0.0001	<0.0005	0.001
W7	6/23/2006			<0.0001	0.011	0.0005	0.067	0.002	<0.0001	<0.0005	31.8	<0.00001	0.0001	0.0007	<0.001
W7	6/28/2006			<0.0001	0.407	0.0008	0.079	0.002	<0.0001	<0.0005	22.8	0.00003	0.0009	0.0015	0.003
W7	7/7/2006			<0.0001	<0.005	0.0006	0.067	0.005	<0.0001	0.0007	26.5	<0.00001	<0.0001	0.0011	0.001
W7	7/12/2006			<0.0001	0.006	0.0005	0.064	0.005	<0.0001	0.0007	26.9	<0.00001	<0.0001	0.001	0.001
W7	7/20/2006			<0.0001	0.006	0.0004	0.068	0.004	<0.0001	<0.0005	33	<0.00001	<0.0001	0.0007	<0.001
W7	7/26/2006														
W7	7/26/2006			<0.0001	0.008	0.0004	0.072	<0.002	<0.0001	<0.0005	33.4	<0.00001	<0.0001	<0.0005	0.002
W7	8/2/2006			<0.0001	0.008	<0.0002	0.07	0.004	<0.0001	<0.0005	33.3	<0.00001	<0.0001	<0.0005	0.001
W7	8/10/2006			<0.0001	0.009	0.0005	0.073	0.005	<0.0001	<0.0005	32.9	<0.00001	<0.0001	0.0007	0.001
W7	8/25/2006														
W7	8/25/2006			<0.0001	0.012	0.0004	0.071	0.005	<0.0001	<0.0005	32.4	<0.00001	0.0002	0.0006	0.001
W7	8/30/2006			<0.0001	0.013	0.0004	0.067	0.004	<0.0001	<0.0005	33.6	<0.00001	0.0001	0.0008	0.002
W7	9/6/2006			<0.0001	0.005	0.0005	0.068	0.005	<0.0001	<0.0005	33.5	<0.00001	<0.0001	0.0008	0.002
W7	9/13/2006			<0.0001	0.008	0.0003	0.064	0.005	<0.0001	<0.0005	32.3	<0.00001	<0.0001	0.0009	0.001
W7	9/20/2006			<0.0001	0.007	0.0005	0.069	0.005	<0.0001	<0.0005	33.5	0.00028	<0.0001	0.0007	0.001
W7	9/28/2006			<0.0001	0.011	0.0004	0.071	0.003	<0.0001	<0.0005	33.3	<0.00001	<0.0001	0.0008	0.001
W7	10/4/2006			<0.0001	<0.005	0.0003	0.068	0.004	<0.0001	<0.0005	32.6	<0.00001	<0.0001	<0.0005	<0.001
W7	10/12/2006			<0.0001	0.007	0.0004	0.067	0.005	<0.0001	<0.0005	34	<0.00001	<0.0001	<0.0005	0.001
W7	6/5/2007														
W7	6/5/2007														
W7	6/5/2007														
W7	6/20/2007			<0.0001	0.014	0.0004	0.06	0.005	<0.0001	<0.0005	29.6	<0.00001	<0.0001	0.0006	0.002
W7	7/18/2007			<0.0001	0.011	0.0005	0.066	0.005	<0.0001	<0.0005	32	<0.00001	<0.0001	0.001	0.002
W7	7/18/2007														
W7	7/18/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	9/18/2007			<0.0001	0.007	0.0004	0.059	0.004	<0.0001	<0.0005	26.9	<0.00001	<0.0001	0.0016	0.001
W7	10/30/2007														
W7	10/30/2007														
W7	10/30/2007														
W7	10/30/2007														
W7	4/22/2008			<0.0010	<0.02	<0.0002	0.075	<0.005	<0.00004		40.9	<0.00008	0.00008	0.0012	<0.001
W7	4/22/2008			<0.0010	<0.02	0.0004	0.08	<0.005	<0.00004		42.2	<0.00008	0.00012	0.0014	0.002
W7	4/22/2008			<0.0010	<0.02	0.0003	0.073	<0.005	<0.00004		39.9	<0.00008	0.00023	<0.0006	0.001
W7	4/22/2008														
W7	6/3/2008														
W7	6/3/2008														
W7	6/3/2008														
W7	6/3/2008			<0.00001	<0.01	<0.0002	0.05	<0.004	<0.00004		23.2	<0.00008	0.00011	0.0013	<0.001
W7	6/17/2008			<0.00001	0.02	<0.0002	0.056	0.006	<0.00004			<0.00008	0.00014	0.0013	0.001
W7	8/6/2008														
W7	8/6/2008														
W7	8/6/2008														
W7	8/6/2008			0.00006	0.014	0.0004	0.068	0.004	<0.0001	<0.0005	31.8	<0.00001	0.0001	0.0006	0.002
W7	9/18/2008			<0.00001	0.07	0.0003	0.043	<0.004	<0.00004		20	<0.00001	0.00025	0.0016	0.004
W7	10/28/2008														
W7	10/28/2008														
W7	10/28/2008														
W7	10/28/2008			<0.00001	0.009	0.0005	0.061	0.006	<0.00004		30.5	<0.00001	0.0001	0.0005	0.001
W7	4/30/2009			<0.00001	0.053	<0.0002	0.01	<0.004	<0.00004	<0.0001	5.23	0.00008	0.0001	0.0006	0.005
W7	6/8/2009			<0.00001	0.013	0.0003	0.052	<0.004	<0.00004	<0.0001	23.8	<0.00001	0.00013	<0.0004	0.002
W7	7/28/2009														
W7	8/6/2009														
W7	8/24/2009														
W7	8/25/2009														
W7	8/31/2009			<0.00001	0.01	0.0004	0.056	<0.004	<0.00004	<0.001	29.1	<0.00001	0.0001	0.0012	0.004
W7	10/7/2009														
W7	10/7/2009														
W7	10/25/2009			<0.00002	0.007	0.0003	0.067	<0.05	<0.0001	<0.001	32.6	0.00004	<0.0005	<0.001	0.0012
W7	10/25/2009														
W7	4/19/2010			<0.00001	0.019	0.0003	0.039	<0.004	<0.00004	<0.001	19.9	0.00002	0.00017	0.0004	0.006
W7	4/27/2010			<0.00002	0.075	0.0003	0.022	<0.05	<0.0001	<0.001	9.79	<0.00001	<0.0005	<0.001	0.0041
W7	4/29/2010			<0.00002	0.065	0.0003	0.023	<0.05	<0.0001	<0.001	9.79	<0.00001	<0.0005	<0.001	0.0034
W7	5/1/2010			<0.00002	0.053	0.0004	0.025	<0.05	<0.0001	<0.001	11	0.00002	<0.0005	<0.001	0.0032
W7	5/3/2010			<0.00002	0.047	0.0004	0.027	<0.05	<0.0001	<0.001	12	<0.00001	<0.0005	<0.001	0.0027
W7	5/5/2010			<0.00002	0.038	0.0005	0.03	<0.05	<0.0001	<0.001	12.9	<0.00001	<0.0005	<0.001	0.0025
W7	5/7/2010			<0.00002	0.028	0.0004	0.034	<0.05	<0.0001	<0.001	14.7	0.00001	<0.0005	<0.001	0.0026
W7	5/9/2010			<0.00002	0.023	0.0004	0.044	<0.05	<0.0001	<0.001	20.5	<0.00001	<0.0005	<0.001	0.0018
W7	5/10/2010			<0.00002	0.021	0.0004	0.046	<0.05	<0.0001	<0.001	20.7	0.00002	<0.0005	<0.001	0.0022
W7	5/12/2010														
W7	5/12/2010			<0.00002	0.017	0.0004	0.046	<0.05	<0.0001	<0.001	21.1	<0.00001	<0.0005	<0.001	0.0016
W7	5/14/2010			<0.00002	0.096	0.0004	0.044	<0.05	<0.0001	<0.001	18.7	0.00001	<0.0005	<0.001	0.0031
W7	5/16/2010			<0.00002	0.012	0.0003	0.045	<0.05	<0.0001	<0.001	22.3	<0.00001	<0.0005	<0.001	0.0013
W7	5/17/2010			<0.00002	0.018	0.0004	0.043	<0.05	<0.0001	<0.001	20.9	<0.00001	<0.0005	<0.001	0.0012
W7	5/19/2010			<0.00002	0.016	0.0003	0.049	<0.05	<0.0001	<0.001	24.4	<0.00001	<0.0005	<0.001	0.0017
W7	5/21/2010			<0.0001	0.013	<0.0004	0.05	<0.1	<0.0002	<0.001	24	<0.0001	<0.0005	<0.002	0.001
W7	5/23/2010			<0.0001	0.017	<0.0004	0.053	<0.1	<0.0002	<0.001	25	<0.0001	<0.0005	<0.002	0.001
W7	5/25/2010			<0.0001	0.024	<0.001	0.065	<0.1	<0.0005	<0.001	30	<0.0001	<0.0005	<0.002	0.003
W7	5/27/2010			<0.0001	0.028	<0.001	0.055	<0.1	<0.0005	<0.001	27	<0.0001	<0.0005	<0.002	0.003
W7	5/29/2010			<0.0001	0.017	0.0005	0.056	<0.1	<0.0002	<0.001	29	<0.0001	<0.0005	<0.002	0.002
W7	5/31/2010			<0.0001	0.02	<0.0004	0.058	<0.1	<0.0002	<0.001	27	<0.0001	<0.0005	<0.002	0.002
W7	6/2/2010			<0.0001	0.012	0.0004	0.058	<0.1	<0.0002	<0.001	29	<0.0001	<0.0005	<0.002	0.001
W7	6/4/2010			<0.0001	0.013	0.0004	0.058	<0.1	<0.0002	<0.001	29	<0.0001	<0.0005	<0.002	0.001
W7	6/6/2010			<0.0001	<0.01	<0.0004	0.062	<0.1	<0.0002	<0.001	33	<0.0001	<0.0005	<0.002	0.001
W7	6/14/2010			<0.0001	0.011	<0.0004	0.064	<0.1	<0.0002	<0.001	31	<0.0001	<0.0005	<0.002	<0.001
W7	6/16/2010			<0.0001	<0.01	<0.0004	0.064	<0.1	<0.0002	<0.001	30	<0.0001	<0.0005	<0.002	<0.001
W7	6/18/2010			<0.0001	<0.01	<0.0004	0.064	<0.1	<0.0002	<0.001	30	<0.0001	<0.0005	<0.002	<0.001
W7	6/20/2010			<0.0001	0.014	<0.0004	0.06	<0.1	<0.0002	<0.001	28	<0.0001	<0.0005	<0.002	0.001
W7	6/22/2010			<0.0001	0.033	0.0005	0.055	<0.1	<0.0002	<0.001	25	<0.0001	<0.0005	<0.002	0.002
W7	6/24/2010			<0.0001	0.018	<0.0004	0.052	<0.1	<0.0002	<0.001	25	<0.0001	<0.0005	<0.002	0.001
W7	6/26/2010			<0.0001	0.012	0.0004	0.06	<0.1	<0.0002	<0.001	26	<0.0001	<0.0005	<0.002	0.001
W7	6/28/2010			<0.0001	<0.01	<0.0004	0.061	<0.1	<0.0002	<0.001	29	<0.0001	<0.0005	<0.002	0.001
W7	7/5/2010			<0.0001	0.019	<0.0004	0.054	<0.1	<0.0002	<0.001	23	<0.0001	<0.0005	<0.002	0.002
W7	8/5/2010			<0.0001	0.011	0.0004	0.065	<0.1	<0.0002	<0.001	30	<0.0001	<0.0005	<0.002	0.002
W7	9/12/2010														
W7	9/12/2010			<0.0001	0.016	0.0005	0.055	<0.1	<0.0002	<0.001	28	<0.0001	<0.0005	<0.002	0.001
W7	10/8/2010			<0.00002	0.011	0.0004	0.06	<0.05	<0.0001	<0.001	29.2	<0.00001	<0.0005	<0.001	0.0012
W7	4/27/2011			<0.00002	0.057	0.0003	0.024	<0.05	<0.0001	<0.001	11.2	0.00003	<0.0005	<0.001	0.0077
W7	5/2/2011			<0.00002	0.076	0.0002	0.02	<0.05	<0.0001	<0.001	8.48	0.00004	<0.0005	<0.001	0.007
W7	6/1/2011			<0.00002	0.013	0.0003	0.056	<0.05	<0.0001	<0.001	23.5	<0.00001	<0.0005	<0.001	0.0016
W7	6/16/2011			<0.00002	0.022	0.0004	0.053	<0.05	<0.0001	<0.001	24.9	<0.00001	<0.0005	<0.001	0.0019
W7	7/1/2011			<0.00002	0.034	0.0007	0.053	<0.05	<0.0001	<0.001	20.6	0.00001	<0.0005	<0.001	0.0031

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Cd-D-IH (mg/L)	Se-D-IH (mg/L)	Ag-D (mg/L)	Dissolved Al (mg/L)	Dissolved As (mg/L)	Dissolved Ba (mg/L)	Dissolved B (mg/L)	Dissolved Be (mg/L)	Dissolved Bi (mg/L)	Dissolved Ca (mg/L)	Dissolved Cd (mg/L)	Dissolved Co (mg/L)	Dissolved Cr (mg/L)	Dissolved Cu (mg/L)
W7	8/2/2011			<0.00002	0.051	0.0007	0.048	<0.05	<0.0001	<0.001	15.6	0.00002	<0.0005	<0.001	0.0041
W7	8/16/2011			<0.00002	0.036	0.0007	0.057	<0.05	<0.0001	<0.001	26	<0.00001	<0.0005	<0.001	0.0019
W7	8/19/2011			<0.00002	0.033	0.0008	0.053	<0.05	<0.0001	<0.001	24.2	0.00001	<0.0005	<0.001	0.002
W7	9/12/2011			<0.00002	0.024	0.0007	0.061	<0.05	<0.0001	<0.001	30.8	0.00001	<0.0005	<0.001	0.0016
W7	10/6/2011			<0.00002	0.013	0.0006	0.057	<0.05	<0.0001	<0.001	27.9	0.00003	<0.0005	<0.001	0.0012
W7	11/22/2011			<0.000020	0.0076	0.0004	0.075	<0.05	<0.00010	<0.0010	38.3	<0.000010	<0.00050	<0.0010	0.00109
W7	1/24/2012			<0.000020	0.0053	0.00041	0.0882	<0.05	<0.00010	<0.0010	43	0.000035	<0.00050	<0.0010	0.00114
W7	2/25/2012			<0.000020	0.0211	0.00033	0.0796	<0.05	<0.00010	<0.0010	41.2	0.000064	<0.00050	<0.0010	0.0154
W7	3/5/2012			<0.000020	0.0031	0.00039	0.0803	<0.05	<0.00010	<0.0010	41.2	<0.000010	<0.00050	<0.0010	0.00122
W7	4/24/2012			<0.000020	0.0499	0.00048	0.0319	<0.05	<0.00010	<0.0010	10.2	0.000023	<0.00050	<0.0010	0.00523
W7	5/6/2012			<0.000020	0.0298	0.00043	0.0319	<0.05	<0.00010	<0.0010	12	0.000013	<0.00050	<0.0010	0.00303
W7	5/6/2012			<0.000020	0.0442	0.00044	0.0324	<0.05	<0.00010	<0.0010	11.9	0.000016	<0.00050	<0.0010	0.00273
W7	6/19/2012			<0.000020	0.0265	0.0007	0.0543	<0.05	<0.00010	<0.0010	23.5	<0.000010	<0.00050	<0.0010	0.00163
W7	7/17/2012			<0.000020	0.0197	0.00086	0.0706	<0.05	<0.00010	<0.0010	33	0.000012	<0.00050	<0.0010	0.00127
W7	7/30/2012			<0.000020	0.0171	0.00095	0.0727	<0.05	<0.00010	<0.0010	33.9	<0.000010	<0.00050	<0.0010	0.00123
W7	8/8/2012			<0.000020	0.0184	0.001	0.0769	<0.05	<0.00010	<0.0010	35.9	<0.000010	<0.00050	<0.0010	0.00112
W7	8/23/2012			<0.000020	0.0178	0.00088	0.0756	<0.05	<0.00010	<0.0010	35.7	<0.000010	<0.00050	<0.0010	0.0017
W7	9/13/2012			<0.000020	0.0239	0.00074	0.0599	<0.05	<0.00010	<0.0010	30.1	<0.000010	<0.00050	<0.0010	0.00244
W7	10/16/2012			<0.000020	0.0114	0.0006	0.0633	<0.05	<0.00010	<0.0010	33.2	<0.000010	<0.00050	<0.0010	0.00094
W7	11/2/2012			<0.000020	0.0095	0.00053	0.0862	<0.05	<0.00010	<0.0010	39.3	0.000031	<0.00050	<0.0010	0.00236
W7	12/29/2012			<0.000020	0.0061	0.00051	0.119	<0.05	<0.00010	<0.0010	56.1	<0.000010	<0.00050	<0.0010	0.00232
W7	4/30/2013			<0.000020	0.0031	0.00025	0.101	<0.05	<0.00010	<0.0010	46.8	<0.000010	<0.00050	<0.0010	0.00108
W7	5/14/2013			<0.000020	0.0436	0.00029	0.023	<0.05	<0.00010	<0.0010	8.81	0.00001	<0.00050	<0.0010	0.00365
W7	6/19/2013			<0.000020	0.0125	0.00052	0.0665	<0.05	<0.00010	<0.0010	29.9	<0.000010	<0.00050	<0.0010	0.00116
W7	6/19/2013			<0.000020	0.0134	0.00051	0.0664	<0.05	<0.00010	<0.0010	30	<0.000010	<0.00050	<0.0010	0.00124
W7	7/13/2013			<0.000020	0.0138	0.00053	0.072	<0.05	<0.00010	<0.0010	34.9	<0.000010	<0.00050	<0.0010	0.00121
W7	8/21/2013			<0.000020	0.0122	0.00056	0.0737	<0.05	<0.00010	<0.0010	35.2	<0.000010	<0.00050	<0.0010	0.00101
W7	9/6/2013			<0.000020	0.016	0.00081	0.0718	<0.05	<0.00010	<0.0010	34.9	<0.000010	<0.00050	<0.0010	0.00113
W7	10/21/2013			<0.000020	0.0093	0.00067	0.0715	<0.05	<0.00010	<0.0010	38	<0.000010	<0.00050	<0.0010	0.00111
W7	11/16/2013			<0.000020	0.0053	0.00048	0.08	<0.05	<0.00010	<0.0010	39.4	<0.000010	<0.00050	<0.0010	0.0008
W7	12/7/2013			<0.000020	0.0035	0.00053	0.0919	<0.05	<0.00010	<0.0010	43.3	<0.000010	<0.00050	<0.0010	0.00078
C4	5/6/2012			<0.000020	0.0622	0.00104	0.0488	<0.05	<0.00010	<0.0010	12	0.000029	0.00071	<0.0010	0.00314
C4	8/1/2012			<0.000020	0.059	0.00184	0.0984	<0.05	<0.00010	<0.0010	31.9	<0.000010	0.00111	<0.0010	0.00181
C4	8/8/2012			<0.000020	0.0425	0.00203	0.0987	<0.05	<0.00010	<0.0010	33.7	<0.000010	0.00108	<0.0010	0.00132
C4	8/25/2012			<0.000020	0.04	0.00197	0.098	<0.05	<0.00010	<0.0010	35	<0.000010	0.00113	<0.0010	0.00176
C4	6/22/2013			<0.000020	0.0342	0.00225	0.075	<0.05	<0.00010	<0.0010	35	<0.000010	0.00063	<0.0010	0.0017
C4	7/19/2013			<0.000020	0.0317	0.00207	0.0699	<0.05	<0.00010	<0.0010	36	<0.000010	0.00062	<0.0010	0.00137
C4	8/22/2013			<0.000020	0.0455	0.00179	0.0794	<0.05	<0.00010	<0.0010	33.1	<0.000010	0.00073	<0.0010	0.00143
C4	9/6/2013			<0.000020	0.0405	0.00196	0.0798	<0.05	<0.00010	<0.0010	34.2	<0.000010	0.00092	<0.0010	0.00161
C4	10/18/2013			<0.000020	0.0221	0.00106	0.0735	<0.05	<0.00010	<0.0010	34.2	<0.000010	0.00079	<0.0010	0.00112
C4	10/18/2013			<0.000020	0.0249	0.00107	0.0732	<0.05	<0.00010	<0.0010	34.5	<0.000010	0.00074	<0.0010	0.00118
C4	11/17/2013			<0.000020	0.033	0.00559	0.169	<0.05	<0.00010	<0.0010	72.1	0.000022	0.00611	0.0013	0.00095
C10	5/12/2012			<0.000020	0.0583	0.0012	0.0501	<0.05	<0.00010	<0.0010	19.9	0.000018	0.00054	<0.0010	0.00212
C10	7/30/2012			<0.000020	0.0571	0.00142	0.0795	<0.05	<0.00010	<0.0010	38.8	<0.000010	<0.00050	<0.0010	0.00198
C10	7/30/2012														
C10	8/10/2012			<0.000020	0.0505	0.00134	0.0786	<0.05	<0.00010	<0.0010	39.8	<0.000010	<0.00050	<0.0010	0.00189
C10	8/27/2012			<0.000020	0.0262	0.00149	0.0804	<0.05	<0.00010	<0.0010	41.8	<0.000010	<0.00050	<0.0010	0.00148
C10	6/22/2013			<0.000020	0.0785	0.00123	0.0721	<0.05	<0.00010	<0.0010	45.1	0.000019	<0.00050	<0.0010	0.00313
C10	7/23/2013			<0.000020	0.0252	0.00116	0.0668	<0.05	<0.00010	<0.0010	37.4	<0.000010	<0.00050	<0.0010	0.00135
C10	8/22/2013			<0.000020	0.0197	0.00141	0.0724	<0.05	<0.00010	<0.0010	48	<0.000010	<0.00050	<0.0010	0.00134
C10	9/8/2013			<0.000020	0.0254	0.00166	0.0764	<0.05	<0.00010	<0.0010	45	<0.000010	0.00058	<0.0010	0.00125
C10	9/8/2013			<0.000020	0.0241	0.00158	0.0763	<0.05	<0.00010	<0.0010	44.8	0.00001	0.0006	0.0017	0.00133
C10	10/18/2013			<0.000020	0.0108	0.00089	0.0599	<0.05	<0.00010	<0.0010	40.1	<0.000010	<0.00050	<0.0010	0.00091

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Dissolved Fe (mg/L)	Dissolved Hg (mg/L)	Dissolved K (mg/L)	Dissolved Li (mg/L)	Dissolved Mg (mg/L)	Dissolved Mn (mg/L)	Dissolved Mo (mg/L)	Dissolved Na (mg/L)	Dissolved Ni (mg/L)	Dissolved Pb (mg/L)	Dissolved P (mg/L)	Dissolved Sb (mg/L)	Dissolved S (mg/L)	Dissolved Se (mg/L)
W1	5/27/2005	0.172	<0.000020	<2.0	<0.0050	8.46	0.00579	<0.0010	4.9	0.0015	<0.00050		<0.00050		<0.0010
W1	6/30/2005	0.046		<2.0	<0.0050	12.6	0.00391	0.00106	8.8	0.00096	<0.000050	<0.30	<0.00010		<0.0010
W1	7/28/2005	0.056		<2.0	<0.0050	13.6	0.0119	0.00131	9.2	0.0011	0.000073	<0.30	<0.00010		<0.0010
W1	8/29/2005	0.25	<0.000020	<2.0	<0.0050	11.7	0.0157	0.001	6.9	0.0014	<0.00050		<0.00050		<0.0010
W1	9/28/2005	0.331	<0.000020	<2.0	<0.0050	11.2	0.0355	<0.0010	6.2	0.0016	<0.00050		<0.00050		<0.0010
W1	10/15/2005	0.266		<2.0	<0.0050	11.7	0.0228	0.000892	6.4	0.0014	<0.000050	<0.30	<0.00010		<0.0010
W2	5/27/2005	0.164	<0.000020	<2.0	<0.0050	8.72	0.00507	<0.0010	5	0.0015	<0.00050		<0.00050		<0.0010
W2	5/27/2005					5.19	0.00345								
W2	6/30/2005	0.039		<2.0	<0.0050	12.3	0.00684	0.00108	8.6	0.00102	<0.000050	<0.30	<0.00010		<0.0010
W2	7/28/2005	0.05		<2.0	<0.0050	13.2	0.014	0.00127	8.7	0.00112	<0.000050	<0.30	<0.00010		<0.0010
W2	8/30/2005	0.234	<0.000020	<2.0	<0.0050	11.7	0.0152	0.001	7	0.0015	<0.00050		<0.00050		<0.0010
W2	9/28/2005	0.338	<0.000020	<2.0	<0.0050	11.2	0.0341	<0.0010	6.2	0.0017	<0.00050		<0.00050		<0.0010
W2	10/15/2005	0.246		<2.0	<0.0050	11.5	0.0208	0.000859	6.3	0.00141	<0.000050	<0.30	<0.00010		<0.0010
W3	5/27/2005	0.116	<0.000020	<2.0	<0.0050	9.48	0.0094	<0.0010	5.2	0.0014	<0.00050		<0.00050		<0.0010
W3	6/30/2005	0.042		<2.0	<0.0050	17.4	0.0052	0.00163	11.7	0.00078	0.000241	<0.30	<0.00010		<0.0010
W3	7/29/2005	0.134		<2.0	<0.0050	17.3	0.0311	0.00158	9.7	0.00115	<0.000050	<0.30	<0.00010		<0.0010
W3	8/29/2005	0.306	<0.000020	<2.0	<0.0050	13.6	0.0657	0.0013	8	0.0014	<0.00050		<0.00050		<0.0010
W3	9/28/2005	0.421	<0.000020	<2.0	<0.0050	12.8	0.102	0.0011	7.7	0.0014	<0.00050		<0.00050		<0.0010
W3	10/15/2005	0.39		<2.0	<0.0050	13.4	0.112	0.00122	7.5	0.00121	<0.000050	<0.30	<0.00010		<0.0010
W8	5/27/2005	0.194	<0.000020	<2.0	<0.0050	5.05	0.00284	<0.0010	2.6	0.0018	<0.00050		<0.00050		<0.0010
W8	6/30/2005	0.249		<2.0	<0.0050	7.68	0.0142	0.000547	4.5	0.00144	<0.000050	<0.30	<0.00010		<0.0010
W8	7/29/2005	0.352		<2.0	<0.0050	9.39	0.0206	0.000589	3.9	0.00157	<0.000050	<0.30	<0.00010		<0.0010
W8	8/30/2005	0.544	<0.000020	<2.0	<0.0050	8.28	0.0296	<0.0010	3.9	0.0016	<0.00050		<0.00050		<0.0010
W8	9/28/2005	0.556	<0.000020	<2.0	<0.0050	7.73	0.0387	<0.0010	3.9	0.0015	<0.00050		<0.00050		<0.0010
W8	10/15/2005	0.711		<2.0	<0.0050	7.92	0.223	0.00169	4.8	0.00096	<0.000050	<0.30	<0.00010		<0.0010
W9	5/27/2005	0.251	0.000038	<2.0	<0.0050	4.18	0.00696	<0.0010	2.3	0.0013	<0.00050		<0.00050		<0.0010
W9	6/30/2005	0.53		<2.0	<0.0050	6.05	0.0378	0.000365	4.1	0.00119	<0.000050	<0.30	<0.00010		<0.0010
W9	7/29/2005	0.647		<2.0	<0.0050	7.46	0.057	0.00108	4	0.00132	0.000185	<0.30	<0.00010		<0.0010
W9	8/29/2005	0.67	<0.000020	<2.0	<0.0050	6.3	0.0606	<0.0010	3.6	0.0012	<0.00050		<0.00050		<0.0010
W9	9/28/2005	0.602	<0.000020	<2.0	<0.0050	5.69	0.0451	<0.0010	3.2	0.0011	<0.00050		<0.00050		<0.0010
W9	10/15/2005	0.438		<2.0	<0.0050	5.55	0.0335	0.000313	3.1	0.00096	<0.000050	<0.30	<0.00010		<0.0010
W10	5/27/2005	0.106	<0.000020	<2.0	<0.0050	3.31	0.00042	<0.0010	<2.0	0.0016	<0.00050		<0.00050		<0.0010
W10	5/27/2005	<0.03	<0.00002	<2.0	<0.005		0.00271	<0.001	<2	<0.001	<0.0005		<0.0005		<0.001
W10	6/30/2005	0.401		<2.0	<0.0050	3.68	0.0299	0.000144	3.5	0.00179	0.000082	<0.30	<0.00010		<0.0010
W10	7/29/2005	0.099		<2.0	<0.0050	4.39	0.00483	0.000137	4.7	0.00123	0.000112	<0.30	<0.00010		<0.0010
W10	8/29/2005	0.081	<0.000020	<2.0	<0.0050	5.95	0.00447	<0.0010	5.4	0.0016	<0.00050		<0.00050		<0.0010
W10	9/28/2005	0.048	<0.000020	<2.0	<0.0050	8.93	0.00069	<0.0010	5.4	<0.0010	<0.00050		<0.00050		<0.0010
W10	10/15/2005	0.055		<2.0	<0.0050	8.98	0.000385	0.000241	4.7	0.00073	<0.000050	<0.30	<0.00010		<0.0010
W6	5/27/2005	0.072	<0.000020	<2.0	<0.0050	5.01	0.00543	<0.0010	2.8	0.0015	<0.00050		<0.00050		<0.0010
W6	6/30/2005	<0.030		<2.0	<0.0050	6.72	0.00324	0.000448	4.8	0.00099	<0.000050	<0.30	<0.00010		<0.0010
W6	7/29/2005	0.062		<2.0	<0.0050	13.2	0.0145	0.000818	7.3	0.0006	0.000083	<0.30	<0.00010		<0.0010
W6	8/30/2005	0.064	<0.000020	<2.0	<0.0050	8.1	0.0211	<0.0010	4.6	0.0011	<0.00050		<0.00050		<0.0010
W6	9/28/2005	0.1	<0.000020	<2.0	<0.0050	8.25	0.0245	<0.0010	4.4	0.0012	<0.00050		<0.00050		<0.0010
W6	10/15/2005	0.088		<2.0	<0.0050	7.66	0.0162	0.000395	4.5	0.00118	<0.000050	<0.30	<0.00010		<0.0010
W6	6/2/2006	0.096		<2	<0.005	4.95	0.00673	0.000305	3.3	0.00182	<0.00005	<0.3	<0.0001		<0.001
W6	8/25/2006					8.7									
W6	9/6/2006	0.11		0.6	0.001	7.7	0.046	<0.001	4.8	0.0006	<0.0001		<0.0002	0.9	<0.0002
W6	6/20/2007														
W6	9/18/2007														
W6	6/17/2008														
W6	9/18/2008	0.09	<0.00001	0.4	<0.001	6.33	0.0099	0.00027	4.27	0.002	<0.0001	0.02	0.0008		0.0012
W6	4/14/2009	0.06	<0.00001	7.4	0.004	45.8	0.0307	0.00312	35.1	<0.001	0.0002	0.13	0.003	34.5	0.0011
W6	4/30/2009	0.12	0.00001	1.9	<0.001	4.9	0.0147	0.00042	3	0.001	0.0003	0.05	0.0006	2.5	<0.0006
W6	7/28/2009														
W6	8/6/2009														
W6	8/6/2009														
W6	8/18/2009	0.02	<0.00001	3.5	0.002	14.2	0.0808	0.0098	14.9	0.001	0.0005	0.01	0.0013	16.6	0.0012
W6	8/25/2009														
W6	10/7/2009														
W6	10/7/2009														
W6	10/25/2009	0.01	<0.00002	3.34	<0.005	15.2	0.215	0.015	17.3	<0.001	<0.0002		0.0006	24	0.002
W6	4/19/2010	0.06	<0.00001	2.3	0.002	13.2	0.0461	0.0017	8.6	<0.001	0.0002	0.02	0.0011	10.1	<0.0006
W6	5/16/2010	0.01	<0.0002	1.39	<0.005	8.38	<0.001	<0.001	4.49	<0.001	<0.0002	<0.01	<0.0005	<3	0.0002
W6	5/17/2010	0.016	<0.0002	1	<0.005	5.28	<0.001	<0.001	3.32	<0.001	<0.0002	0.016	<0.0005	<3	<0.0001
W6	5/19/2010	0.021	<0.00002	1.19	<0.005	7.41	<0.001	<0.001	5.01	<0.001	<0.0002	0.013	<0.0005	5	0.0001
W6	5/21/2010	<0.02	<0.0002	<1	<0.01	5	<0.001	<0.001	3	<0.001	<0.0002		<0.0005	<60	<0.0008
W6	5/23/2010	0.029	<0.0002	1	<0.01	8	<0.001	<0.001	6	<0.001	<0.0002		<0.0005	<60	<0.0008
W6	5/25/2010	0.024	<0.00002	1	<0.01	9	<0.001	<0.001	8	<0.003	<0.0002		<0.0005	<60	<0.001
W6	5/29/2010	0.03	<0.00002	<1	<0.01	4	<0.001	<0.001	2	<0.001	<0.0002		<0.0005	<60	<0.0008
W6	6/2/2010	<0.02	<0.0002	2	<0.01	20	0.01	0.003	13	<0.001	<0.0002		<0.0005	<60	<0.0008
W6	6/4/2010	0.027	<0.0002	<1	<0.01	6	<0.001	<0.001	3	<0.001	<0.0002		<0.0005	<60	<0.0008
W6	6/6/2010	0.027		<1	<0.01	7	<0.001	<0.001	4	0.001	<0.0002		<0.0005	<60	<0.0008
W6	6/14/2010	0.021	<0.0002	<1	<0.01	7	<0.001	<0.001	5	0.001	<0.0002		<0.0005	<60	<0.0008
W6	6/16/2010	0.021	<0.0002	<1	<0.01	7	<0.001	<0.001	4	<0.001	<0.0002		<0.0005	<60	<0.0008
W6	6/18/2010	0.021	<0.0002	<1	<0.01	7	<0.001	<0.001	4	<0.001	<0.0002		<0.0005	<60	<0.0008
W6	6/20/2010	0.027	<0.0002	<1	<0.01	7	<0.001	<0.001	4	0.001	<0.0002		<0.0005	<60	<0.0008
W6	6/22/2010	0.025	<0.0002	<1	<0.01	7	<0.001	<0.001	4	0.001	<0.0002		<0.0005	<60	<0.0008
W6	6/24/2010	0.025	<0.0002	<1	<0.01	7	<0.001	<0.001	4	<0.001	<0.0002		<0.0005	<60	<0.0008
W6	6/26/2010	0.025	<0.0002	<1	<0.01	7	0.001	<0.001	4	0.001	<0.0002		<0.0005	<60	<0.0008
W6	6/26/2010														
W6	6/28/2010	0.025	<0.0002	<1	<0.01	7	<0.001	<0.001	4	0.001	<0.0002		<0.0005	<60	<0.0008
W6	7/5/2010	0.092	<0.0002	<1	<0.01	6	0.005	<0.001	4	0.001	<0.0002		<0.0005	<60	<0.0008
W6	8/5/2010	0.069	<0.0002	<1	<0.01	8	0.018	<0.001	4	0.001	<0.0002		<0.0005	<60	<0.0008
W6	9/9/2010	0.061	<0.0002	<1	<0.01	8	0.02	<0.001	4	0.001	<0.0002		<0.0005	<60	<0.0008
W6	10/8/2010	0.054	<0.00002	0.42	<0.005	7.52	0.019	<0.001	3.79	0.001	<0.0002		<0.0005	<3	<0.0001
W6	8/19/2011	0.025	<0.00002	1.66	<0.005	22.2	0.019	0.003	15.2	<0.001	<0.0002	<0.01	<0.0005	18	0.0004
W6	7/17/2012	0.0679	<0.000010	0.52	<0.0050	7.16	0.0542	<0.0010	4.31	0.001	<0.00020	<0.01	<0.00050	<3.0	0.00012
W6	7/30/2012	0.0814	<0.000010	0.542	<0.0050	7.02	0.102	<0.0010	4.2	<0.0010	<0.00020	<0.01	<0.00050	<3.0	<0.00010
W6	8/9/2012	0.0845	<0.000010	0.578	<0.0050	7.95	0.134	<0.0010	4.56	0.001	<0.00020	0.011	<0.00050	<3.0	0.0001
W6	8/25/2012	0.0799	<0.000010	0.648	<0.0050	7.34	0.134	<0.0010	4.21	0.001	<0.00020	0.012	<0.00050	<3.0	<0.00010
W6	9/13/2012	0.108	<0.000010	0.411	<0.0050	6.66	0.0408	<0.0010	3.89	0.0014	<0.00020	<0.01	<0.00050	<3.0	<0.00010
W6	10/13/2012		<0.000010												
W6	11/12/2012	0.0235	<0.000010	0.562	<0.0050	11.8	0.0013	<0.0010	6.34	0.0014	<0.00020	<0.01	<0.00050	<3.0	0.0001
W6	5/14/2013	0.183	<0.000010	1.73	<0.0050	2.31	0.0433	<0.0010	0.999	0.0013	<0.00020	0.044	<0.00050	<3.0	<0.00010
W6	6/22/2013	0.0718	<0.000010	0.724	<0.0050	7.04	0.0293	<0.0010	3.95	0.0012	<0.00020	<0.01	<0.00050	<3.0	<0.00010
W6	7/12/2013	0.105	<0.000010	0.482	<0.0050	6.83	0.022	<0.0010	4.07	0.0011	<0.00020	<0.01	<0.00050	<3.0	<0.00010
W6	8/22/2013	0.103	<0.000010	0.489	<0.0050	7.54	0.0286	<0.0010	4.33	0.0011	<0.00020	<0.01	<0.00050	<3.0	<0.00010
W6	9/6/2013	0.111	<0.000010	0.554	<0.0050	7.18	0.0323	<0.0010	4.02	0.0011	<0.00020	0.013	<0.00050	<3.0	<0.00010
W6	10/7/2013	0.145	<0.000010	0.581	<0.0050	7.36	0.048	<0.0010	4.6	0.0013	<0.00020	0.011	<0.00050	<3.0	<0.00010
W6	11/16/2013	0.124	<0.000010	0.589	<0.0050	8.66	0.0399	<0.0010	4.77	0.001	<0.00020	0.015	<0.00050	<3.0	<0.00010
W6	12/7/2013	0.0144	<0.000010	0.562	<0.0050	11.7	0.0017	<0.0010	6.36	0.0011	<0.00020	0.015	<0.00050	<3.0	<0.00010
W7	5/27/2005	0.136	<0.000020	<2.0	<0.0050	6.77	0.0193	<0.0010	3.9	0.0016	<0.00050		<0.00050		<0.0010
W7	6/30/2005	0.07		<2.0	<0.0050	10.2	0.0415	0.00124	6.9	0.00093	<0.000050	<0.30	<0.00010		<0.0010
W7	7/29/2005	0.105		<2.0	<0.0050	15.6	0.0244	0.00127	8.8	0.00095	<0.000050	<0.30	<0.00010		<0.0010
W7	8/30/2005	0.292	<0.000020	<2.0	<0.0050	9.7	0.0315	0.001	5.9	0.0015	<0.00050		<0.00050		<0.0010
W7	9/28/2005	0.32	<0.000020	<2.0	<0.0050	9.32	0.0414	<0.0010	5.3	0.0018	<0.00050		<0.00050		<0.0010
W7	10/15/2005	0.253		<2.0	<0.0050	9.35	0.0383	0.000925	5.5	0.00146	<0.000050	<0.30	<0.00010		<0.0010
W7	6/2/2006	0.546		<2	<0.005	4.85	0.0882	0.000697	3.8	0.0022	0.000054	<0.3	<0.0001		<0.001
W7	6/8/2006														

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Dissolved Fe (mg/L)	Dissolved Hg (mg/L)	Dissolved K (mg/L)	Dissolved Li (mg/L)	Dissolved Mg (mg/L)	Dissolved Mn (mg/L)	Dissolved Mo (mg/L)	Dissolved Na (mg/L)	Dissolved Ni (mg/L)	Dissolved Pb (mg/L)	Dissolved P (mg/L)	Dissolved Sb (mg/L)	Dissolved S (mg/L)	Dissolved Se (mg/L)
W7	6/15/2006														
W7	6/15/2006	0.09		<0.4	<0.001	8.9	0.044	0.001	5.9	0.001	<0.0001		<0.0002	3.5	0.0002
W7	6/23/2006	0.09		1.1	<0.001	10.4	0.043	0.001	5.9	0.0013	<0.0001		<0.0002	4.3	0.0003
W7	6/28/2006	0.06		0.8	<0.001	9.2	0.011	<0.001	6.2	0.0036	0.0005		<0.0002	3.4	<0.0002
W7	7/7/2006	0.06		0.8	0.001	10.7	0.023	0.002	6.7	0.0015	<0.0001		<0.0002	4	<0.0002
W7	7/12/2006	0.06		0.9	0.001	10.5	0.022	0.001	6.6	0.0007	<0.0001		<0.0002	4	0.0002
W7	7/20/2006	0.11		1	0.001	11.7	0.042	0.001	6.7	0.0006	0.0002		<0.0002	4.1	<0.0002
W7	7/26/2006														
W7	7/26/2006	0.1		1	<0.001	10.9	0.044	<0.001	7	0.0021	<0.0001		<0.0002	4.2	<0.0002
W7	8/2/2006	0.07		1.1	<0.001	11	0.033	<0.001	7.2	0.0006	<0.0001		<0.0002	4	<0.0002
W7	8/10/2006	0.08		1	0.001	10.8	0.039	0.001	7.2	0.0006	<0.0001		<0.0002	4	<0.0002
W7	8/25/2006														
W7	8/25/2006	0.08		0.9	0.001	11.7	0.037	0.001	6.7	0.0009	<0.0001		<0.0002	3.8	<0.0002
W7	8/30/2006	0.1		1.2	0.002	11.7	0.038	0.001	7.7	0.0012	<0.0001		<0.0002	3.7	<0.0002
W7	9/6/2006	0.1		0.8	0.002	11.2	0.036	0.001	7.5	0.001	<0.0001		<0.0002	3.7	<0.0002
W7	9/13/2006	0.11		0.9	0.002	11.4	0.03	0.001	7.4	0.0009	<0.0001		<0.0002	3.5	<0.0002
W7	9/20/2006	0.09		1	0.002	11.6	0.04	0.001	6.6	<0.0005	<0.0001		<0.0002	3.9	0.0002
W7	9/28/2006	0.08		1.1	0.001	11.1	0.035	0.001	7.4	0.0008	<0.0001		<0.0002	3.7	<0.0002
W7	10/4/2006	0.11		1.1	0.001	10.9	0.035	0.001	7.3	0.0007	<0.0001		<0.0002	3.7	<0.0002
W7	10/12/2006	0.09		1.1	0.001	11.4	0.043	0.001	7.4	0.0006	<0.0001		<0.0002	3.9	0.0003
W7	6/5/2007														
W7	6/5/2007														
W7	6/5/2007														
W7	6/20/2007	0.05	<0.0001	1	0.001	9.8	0.031	0.001	7.4	<0.0005	<0.0001		<0.0002	4.2	<0.0002
W7	7/18/2007	0.04	<0.0001	0.9	0.001	11.6	0.019	0.001	7.6	0.0007	<0.0001		<0.0002	3.9	<0.0002
W7	7/18/2007														
W7	7/18/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	9/18/2007	0.08	<0.0001	0.9	0.001	9.8	0.018	0.002	7.2	0.0008	<0.0001		<0.0002	2.9	<0.0002
W7	10/30/2007														
W7	10/30/2007														
W7	10/30/2007														
W7	10/30/2007														
W7	4/22/2008	<0.02	<0.00001	2.68	0.002	20.4	0.025	0.00161	12.6	<0.001	<0.0001	<0.01	0.006		<0.0006
W7	4/22/2008	0.07	<0.00001	2.78	0.002	21.2	0.027	0.0017	12.7	0.002	0.0001	0.02	0.0084		<0.0006
W7	4/22/2008	<0.02	<0.00001	2.63	0.002	19.9	0.0242	0.00166	12.4	<0.001	<0.0001	<0.01	0.0016		<0.0006
W7	4/22/2008														
W7	6/3/2008														
W7	6/3/2008														
W7	6/3/2008														
W7	6/3/2008	0.05	<0.00001	0.84	<0.001	8.39	0.0418	0.00124	6.82	0.001	<0.0001		0.0006		<0.0006
W7	6/17/2008	0.07			<0.001		0.0682	0.00149		0.001	<0.0001		0.001		0.0022
W7	8/6/2008														
W7	8/6/2008														
W7	8/6/2008														
W7	8/6/2008	0.07	<0.0001	1	0.001	10.6	0.02	0.001	7.4	0.001	<0.0001	<0.05	0.0014	3.5	<0.0002
W7	9/18/2008	0.2	<0.00001	0.53	<0.001	6.63	0.0198	0.00072	4.92	0.001	0.0017	0.02	0.0008		<0.0006
W7	10/28/2008														
W7	10/28/2008														
W7	10/28/2008														
W7	10/28/2008	0.078	<0.00001	0.8	<0.001	11.4	0.0338	0.00108	6.4	0.001	<0.0001	<0.01	0.0009		<0.0006
W7	4/30/2009	0.07	0.00001	1.5	<0.001	1.68	0.0114	0.00013	1.2	<0.001	0.0001	0.06	0.0007	0.4	<0.0006
W7	6/8/2009	0.05	<0.00001	0.7	<0.001	8.18	0.0129	0.00127	5.3	0.001	<0.0001	0.01	0.0011	2.8	<0.0006
W7	7/28/2009														
W7	8/6/2009														
W7	8/24/2009														
W7	8/25/2009														
W7	8/31/2009	0.1	<0.00001	0.9	<0.001	9.9	0.0128	0.0011	6.8	0.001	0.0004	0.03	0.0009	3	<0.0006
W7	10/7/2009														
W7	10/7/2009														
W7	10/25/2009	0.046	<0.00002	0.95	<0.005	10.8	0.024	0.001	6.68	<0.001	<0.0002		<0.0005	4	0.0002
W7	10/25/2009														
W7	4/19/2010	0.09	<0.00001	2.1	<0.001	7.3	0.0188	0.0008	5	<0.001	0.0003	0.02	0.001	2.7	<0.0006
W7	4/27/2010	0.15	<0.00002	1.04	<0.005	2.87	<0.001	<0.001	1.86	0.001	<0.0002		<0.0005	<3	<0.0001
W7	4/29/2010	0.18	<0.00002	0.9	<0.005	2.86	<0.001	<0.001	1.92	0.001	<0.0002		<0.0005	<3	<0.0001
W7	5/1/2010	0.26	<0.00002	0.71	<0.005	3.26	<0.001	<0.001	2.22	0.001	<0.0002	0.023	<0.0005	<3	<0.0001
W7	5/3/2010	0.378	<0.00002	0.64	<0.005	3.67	<0.001	<0.001	2.6	0.001	<0.0002		<0.0005	<3	<0.0001
W7	5/5/2010	0.574	<0.00002	0.57	<0.005	4.09	<0.001	<0.001	2.85	0.001	<0.0002	0.021	<0.0005	<3	<0.0001
W7	5/7/2010	0.359	<0.00002	0.68	<0.005	4.9	<0.001	<0.001	3.35	0.002	<0.0002	0.017	<0.0005	<3	<0.0001
W7	5/9/2010	0.297	<0.00002	0.78	<0.005	6.97	<0.001	<0.001	4.54	0.001	<0.0002	0.014	<0.0005	<3	0.0001
W7	5/10/2010	0.276	<0.00002	0.8	<0.005	7.08	<0.001	<0.001	4.69	0.001	<0.0002	0.013	<0.0005	3	0.0001
W7	5/12/2010														
W7	5/12/2010	0.173	0.00002	0.77	<0.005	7.06	<0.001	0.001	4.6	0.001	<0.0002	<0.01	<0.0005	<3	0.0001
W7	5/14/2010	0.339	<0.00002	0.71	<0.005	6.27	0.039	<0.001	4.05	0.002	<0.0002		<0.0005	<3	0.0001
W7	5/16/2010	0.08	<0.0002	0.8	<0.005	7.28	<0.001	0.001	4.75	<0.001	<0.0002	<0.01	<0.0005	<3	0.0001
W7	5/17/2010	0.068	<0.0002	0.78	<0.005	6.82	<0.001	0.001	4.36	<0.001	<0.0002	<0.01	<0.0005	<3	0.0001
W7	5/19/2010	0.069	<0.00002	0.85	<0.005	9.16	<0.001	0.001	6.6	<0.001	<0.0002	<0.01	<0.0005	3	0.0003
W7	5/21/2010	0.057	<0.0002	<1	<0.01	8	<0.001	0.001	6	0.001	<0.0002		<0.0005	<60	<0.0008
W7	5/23/2010	0.069	<0.0002	<1	<0.01	9	<0.001	0.001	6	0.001	<0.0002		<0.0005	<60	<0.0008
W7	5/25/2010	0.062	<0.00002	1	<0.01	12	0.001	0.002	8	<0.003	<0.0002		<0.0005	<60	<0.001
W7	5/27/2010	0.069	<0.00002	1	<0.01	11	0.014	0.002	8	<0.003	<0.0002		<0.0005	<60	<0.001
W7	5/29/2010	0.051	<0.00002	1	<0.01	10	0.015	0.002	6	0.001	<0.0002		<0.0005	<60	<0.0008
W7	5/31/2010	0.05	<0.00002	<1	<0.01	10	0.012	0.002	7	<0.001	<0.0002		<0.0005	<60	<0.0008
W7	6/2/2010	0.03	<0.0002	<1	<0.01	10	<0.001	0.002	6	<0.001	<0.0002		<0.0005	<60	<0.0008
W7	6/4/2010	0.03	<0.0002	<1	<0.01	10	0.001	0.002	6	<0.001	<0.0002		<0.0005	<60	<0.0008
W7	6/6/2010	0.024		1	<0.01	11	0.004	0.002	7	<0.001	<0.0002		<0.0005	<60	<0.0008
W7	6/14/2010	0.025	<0.0002	1	<0.01	11	0.002	0.002	8	<0.001	<0.0002		<0.0005	<60	<0.0008
W7	6/16/2010	0.026	<0.0002	1	<0.01	11	0.004	0.002	7	<0.001	<0.0002		<0.0005	<60	<0.0008
W7	6/18/2010	0.023	<0.0002	1	<0.01	11	<0.001	0.002	7	<0.001	<0.0002		<0.0005	<60	<0.0008
W7	6/20/2010	0.028	<0.0002	<1	<0.01	10	0.003	0.002	7	<0.001	<0.0002		<0.0005	<60	<0.0008
W7	6/22/2010	0.082	<0.0002	<1	<0.01	9	0.001	0.001	6	0.001	<0.0002		<0.0005	<60	<0.0008
W7	6/24/2010	0.056	<0.0002	<1	<0.01	9	0.001	0.001	6	0.001	<0.0002		<0.0005	<60	<0.0008
W7	6/26/2010	0.038	<0.0002	<1	<0.01	10	0.003	0.001	7	0.004	<0.0002		<0.0005	<60	<0.0008
W7	6/28/2010	0.028	<0.0002	<1	<0.01	10	0.002	0.001	7	<0.001	<0.0002		<0.0005	<60	<0.0008
W7	7/5/2010	0.104	<0.0002	<1	<0.01	8	0.003	0.001	6	0.001	<0.0002		<0.0005	<60	<0.0008
W7	8/5/2010	0.112	<0.0002	<1	<0.01	10	0.023	0.001	6	0.001	<0.0002		<0.0005	<60	<0.0008
W7	9/12/2010														
W7	9/12/2010	0.231	<0.0002	<1	<0.01	9	0.017	0.001	5	0.002	<0.0002		<0.0005	<60	<0.0008
W7	10/8/2010	0.127	<0.00002	0.71	<0.005	9.34	0.027	0.001	5.34	0.001	<0.0002		<0.0005	3	0.0001
W7	4/27/2011	0.148	<0.00002	1.75	<0.005	3.57	0.027	<0.001	2.12	0.001	<0.0002	0.048	<0.0005	<3	<0.0001
W7	5/2/2011	0.169	<0.00002	1.39	<0.005	2.57	0.021	<0.001	1.43	0.001	<0.0002		<0.0005	<3	<0.0001
W7	6/1/2011	0.099	<0.00002	0.85	<0.005	8.19	0.011	0.001	5.09	0.001	<0.0002	0.011	<0.0005	4	0.0001
W7	6/16/2011	0.207	<0.00002	0.67	<0.005	8.56	0.011	0.001	5.67	0.002	<0.0002	0.014	<0.0005	<3	0.0001
W7	7/1/2011	0.441	0.00005	0.39	<0.005	6.25	0.037	<0.001	4.4	0.002	<0.0002	0.017	<0.0005	<3	0.0001

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Dissolved Fe (mg/L)	Dissolved Hg (mg/L)	Dissolved K (mg/L)	Dissolved Li (mg/L)	Dissolved Mg (mg/L)	Dissolved Mn (mg/L)	Dissolved Mo (mg/L)	Dissolved Na (mg/L)	Dissolved Ni (mg/L)	Dissolved Pb (mg/L)	Dissolved P (mg/L)	Dissolved Sb (mg/L)	Dissolved S (mg/L)	Dissolved Se (mg/L)
W7	8/2/2011	0.614	<0.00002	0.29	<0.005	4.85	0.064	<0.001	3.69	0.003	<0.0002	0.02	<0.0005	<3	0.0001
W7	8/16/2011	0.593	<0.00002	0.46	<0.005	7.96	0.054	<0.001	4.81	0.002	<0.0002	0.023	<0.0005	<3	0.0001
W7	8/19/2011	0.556	<0.00002	0.41	<0.005	7.89	0.06	<0.001	4.58	0.002	<0.0002	0.032	<0.0005	<3	0.0001
W7	9/12/2011	0.616	<0.00002	0.62	<0.005	10.1	0.079	<0.001	5.73	0.002	<0.0002	0.023	<0.0005	<3	0.0001
W7	10/6/2011	0.346	<0.00002	0.62	<0.005	9.89	0.07	0.001	5.68	0.001	<0.0002	0.015	<0.0005	<3	0.0001
W7	11/22/2011	0.167	<0.000020	0.864	<0.0050	12.8	0.0626	0.0013	6.99	0.0012	<0.00020	0.014	<0.00050	4.6	0.00016
W7	1/24/2012	0.0944	<0.000010	1.6	<0.0050	14.9	0.0614	0.0015	8.72	<0.0010	<0.00020	0.012	<0.00050	7.3	0.00043
W7	2/25/2012	0.0488	<0.000010	1.63	<0.0050	15.1	0.007	0.0014	8.86	<0.0010	0.00024	0.031	<0.00050	7.3	0.00034
W7	3/5/2012	0.0124	<0.000010	1.5	<0.0050	14.2	0.0015	0.0011	8.03	<0.0010	<0.00020	<0.01	<0.00050	6.1	0.00031
W7	4/24/2012	0.475	<0.000010	1.18	<0.0050	3.2	0.0486	<0.0010	1.9	0.0012	<0.00020	0.027	<0.00050	<3.0	<0.00010
W7	5/6/2012	0.427	<0.000010	0.932	<0.0050	3.95	0.0108	<0.0010	2.4	0.0011	<0.00020	0.021	<0.00050	<3.0	<0.00010
W7	5/6/2012	0.462	<0.000010	0.948	<0.0050	4.04	0.0113	<0.0010	2.44	0.0012	<0.00020	0.021	<0.00050	<3.0	<0.00010
W7	6/19/2012	0.56	<0.000010	0.583	<0.0050	8.07	0.0433	0.0013	5.29	0.002	<0.00020	0.02	<0.00050	<3.0	0.00015
W7	7/17/2012	0.965	<0.000010	0.791	<0.0050	11	0.0366	0.0015	6.55	0.0016	<0.00020	0.035	<0.00050	<3.0	0.00013
W7	7/30/2012	0.964	<0.000010	0.833	<0.0050	12.1	0.0534	0.0015	6.98	0.0015	<0.00020	0.032	<0.00050	<3.0	0.00016
W7	8/8/2012	1.06	<0.000010	0.882	<0.0050	12.6	0.0554	0.0016	7.09	0.0017	<0.00020	0.033	<0.00050	<3.0	0.00014
W7	8/23/2012	0.978	<0.000010	0.942	<0.0050	13	0.057	0.0016	7.35	0.0016	<0.00020	0.031	<0.00050	3.1	0.00015
W7	9/13/2012	0.734	<0.000010	0.567	<0.0050	9.91	0.0956	<0.0010	5.77	0.0036	<0.00020	0.023	<0.00050	<3.0	0.00014
W7	10/16/2012	0.603	<0.000010	0.675	<0.0050	11.2	0.073	0.0011	6.34	0.0015	<0.00020	0.021	<0.00050	3.4	0.00014
W7	11/2/2012	0.183	<0.000010	1.34	<0.0050	14.6	0.101	0.0013	8.07	0.0012	0.00021	0.031	<0.00050	5.2	0.00018
W7	12/29/2012	0.0301	<0.000010	2.54	<0.0050	20.2	0.0374	0.0012	12.2	0.0011	<0.00020	<0.01	<0.00050	10.9	0.00045
W7	4/30/2013	0.0143	<0.000010	1.92	<0.0050	18.3	0.0528	0.0013	10.1	<0.0010	<0.00020	<0.01	<0.00050	8	0.00043
W7	5/14/2013	0.306	<0.000010	1.03	<0.0050	2.75	0.103	<0.0010	1.52	0.001	<0.00020	0.027	<0.00050	<3.0	<0.00010
W7	6/19/2013	0.285	<0.000010	0.94	<0.0050	9.4	0.0317	0.0015	5.9	0.0013	<0.00020	0.019	<0.00050	3.2	0.00011
W7	6/19/2013	0.291	<0.000010	0.955	<0.0050	9.47	0.0339	0.0016	5.88	0.0012	<0.00020	0.019	<0.00050	3.5	0.00019
W7	7/13/2013	0.416	<0.000010	0.91	<0.0050	11	0.026	0.0015	6.52	0.0013	<0.00020	0.016	<0.00050	3.2	0.00015
W7	8/21/2013	0.3	<0.000010	0.934	<0.0050	12.7	0.0355	0.0015	7.34	0.0012	<0.00020	0.017	<0.00050	3.2	0.00015
W7	9/6/2013	0.663	<0.000010	0.887	<0.0050	12.3	0.081	0.0013	7.31	0.0016	<0.00020	0.026	<0.00050	<3.0	0.00017
W7	10/21/2013	0.549	<0.000010	0.878	<0.0050	12.6	0.1	0.0012	7.03	0.0012	<0.00020	0.025	<0.00050	4.2	0.00017
W7	11/16/2013	0.169	<0.000010	1.15	<0.0050	13.3	0.0809	0.0016	7.71	<0.0010	<0.00020	0.017	<0.00050	5.4	0.00024
W7	12/7/2013	0.109	<0.000010	1.38	<0.0050	14.5	0.143	0.0011	8.17	<0.0010	<0.00020	<0.01	<0.00050	5.3	0.00024
C4	5/6/2012	1.11	<0.000010	0.895	<0.0050	3.4	0.243	<0.0010	2.4	0.0023	<0.00020	0.037	<0.00050	<3.0	<0.00010
C4	8/1/2012	1.5	<0.000010	0.754	<0.0050	9.12	0.622	0.0011	4.95	0.0031	<0.00020	0.049	<0.00050	3	0.00011
C4	8/8/2012	1.6	<0.000010	0.759	<0.0050	9.81	0.683	0.0011	5.16	0.0027	<0.00020	0.047	<0.00050	<3.0	0.00011
C4	8/25/2012	1.68	<0.000010	0.989	<0.0050	10.3	0.682	0.0012	5.21	0.0029	<0.00020	0.06	<0.00050	<3.0	<0.00010
C4	6/22/2013	1.2	<0.000010	0.983	<0.0050	9.29	0.405	0.0018	5.62	0.0027	<0.00020	0.041	<0.00050	3.2	0.00013
C4	7/19/2013	1.56	<0.000010	0.845	<0.0050	9.31	0.385	0.0017	5.66	0.0025	<0.00020	0.04	<0.00050	<3.0	0.00012
C4	8/22/2013	1.44	<0.000010	0.776	<0.0050	9.87	0.448	0.0013	5.49	0.0027	<0.00020	0.043	<0.00050	<3.0	0.00012
C4	9/6/2013	1.78	<0.000010	0.777	<0.0050	9.52	0.495	0.001	5.28	0.0028	<0.00020	0.048	<0.00050	3.1	<0.00010
C4	10/18/2013	1.15	<0.000010	0.754	<0.0050	10.5	0.417	<0.0010	5.46	0.0023	<0.00020	0.033	<0.00050	4.5	<0.00010
C4	10/18/2013	1.16	<0.000010	0.762	<0.0050	10.4	0.408	0.0012	5.38	0.0025	<0.00020	0.036	<0.00050	4.4	<0.00010
C4	11/17/2013	14.2	<0.000010	0.985	<0.0050	18.8	3.09	0.0011	8.06	0.0061	<0.00020	0.117	<0.00050	4.5	0.00024
C10	5/12/2012	1.55	<0.000010	1.08	<0.0050	4.24	0.305	<0.0010	2.95	0.0019	0.00025	0.035	<0.00050	<3.0	0.0001
C10	7/30/2012	1.15	<0.000010	0.921	<0.0050	8.1	0.189	0.0014	5.64	0.0023	0.00067	0.039	<0.00050	<3.0	0.00014
C10	7/30/2012														
C10	8/10/2012	0.883	<0.000010	0.824	<0.0050	8.59	0.252	0.0019	5.34	0.0022	<0.00020	0.034	<0.00050	<3.0	0.0001
C10	8/27/2012	1	<0.000010	1.02	<0.0050	8.82	0.27	0.0011	5.26	0.0023	<0.00020	0.038	<0.00050	<3.0	0.00012
C10	6/22/2013	0.802	<0.000010	1.45	<0.0050	8.63	0.0628	0.0012	6.07	0.0025	<0.00020	0.053	<0.00050	<3.0	0.00011
C10	7/23/2013	0.54	<0.000010	0.848	<0.0050	7.65	0.154	0.001	5.75	0.0022	<0.00020	0.025	<0.00050	<3.0	<0.00010
C10	8/22/2013	0.992	<0.000010	0.979	<0.0050	9.84	0.063	0.0015	6.05	0.0023	<0.00020	0.032	<0.00050	<3.0	<0.00010
C10	9/8/2013	1.26	<0.000010	1.03	<0.0050	9.12	0.137	<0.0010	5.65	0.0025	<0.00020	0.04	<0.00050	<3.0	<0.00010
C10	9/8/2013	1.28	<0.000010	1.03	<0.0050	9.37	0.139	<0.0010	5.79	0.0025	<0.00020	0.043	<0.00050	<3.0	<0.00010
C10	10/18/2013	0.761	<0.000010	0.844	<0.0050	8.73	0.187	<0.0010	5.44	0.0019	<0.00020	0.027	<0.00050	<3.0	<0.00010

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Dissolved Si (mg/L)	Dissolved Sn (mg/L)	Dissolved Sr (mg/L)	Dissolved Te (mg/L)	Dissolved Th (mg/L)	Dissolved Ti (mg/L)	Dissolved Tl (mg/L)	Dissolved U (mg/L)	Dissolved V (mg/L)	Dissolved Zn (mg/L)	Dissolved Zr (mg/L)	P-E (mg/L)	TT Ag (mg/L)	TT Al (mg/L)
W1	5/27/2005		<0.00050				<0.010	<0.00020	0.00049	<0.030	<0.0050			0.000036	0.613
W1	6/30/2005	5.53	<0.00010	0.3			<0.010	<0.00010	0.00113	<0.0010	<0.0010			<0.000010	0.0218
W1	7/28/2005	5.55	<0.00010	0.331			<0.010	<0.00010	0.00133	<0.0010	0.0025			<0.000010	0.0174
W1	8/29/2005		<0.00050				<0.010	<0.00020	0.00077	<0.030	<0.0050			<0.000020	0.0524
W1	9/28/2005		<0.00050				<0.010	<0.00020	0.00055	<0.030	<0.0050			<0.000020	0.127
W1	10/15/2005	6.25	<0.00010	0.257			<0.010	<0.00010	0.000727	<0.0010	0.0014			<0.000010	0.15
W2	5/27/2005		<0.00050				<0.010	<0.00020						<0.000020	0.362
W2	5/27/2005						<0.01		0.00081	<0.03	<0.005				1.14
W2	6/30/2005	5.4	<0.00010	0.307			<0.010	<0.00010	0.0011	<0.0010	0.0015			<0.000010	0.0111
W2	7/28/2005	5.31	<0.00010	0.341			<0.010	<0.00010	0.00127	<0.0010	0.0011			<0.000010	0.0127
W2	8/30/2005		<0.00050				<0.010	<0.00020	0.00077	<0.030	<0.0050			<0.000020	0.0475
W2	9/28/2005		<0.00050				<0.010	<0.00020	0.00055	<0.030	<0.0050			<0.000020	0.133
W2	10/15/2005	6.19	<0.00010	0.256			<0.010	<0.00010	0.000713	<0.0010	<0.0010			<0.000010	0.0972
W3	5/27/2005		<0.00050				<0.010	<0.00020	0.00031	<0.030	<0.0050			<0.000020	0.152
W3	6/30/2005	5.89	<0.00010	0.297			<0.010	<0.00010	0.000903	<0.0010	0.0116			<0.000010	0.0313
W3	7/29/2005	5.56	<0.00010	0.279			<0.010	<0.00010	0.000717	<0.0010	0.0015			<0.000030	0.732
W3	8/29/2005		<0.00050				<0.010	<0.00020	0.00045	<0.030	<0.0050			0.000027	0.267
W3	9/28/2005		<0.00050				<0.010	<0.00020	0.00034	<0.030	<0.0050			0.000029	0.236
W3	10/15/2005	6.16	<0.00010	0.225			<0.010	<0.00010	0.00042	<0.0010	0.0018			<0.000010	0.09
W8	5/27/2005		<0.00050				<0.010	<0.00020	<0.00020	<0.030	<0.0050			<0.000020	0.0506
W8	6/30/2005	5.25	<0.00010	0.113			<0.010	<0.00010	0.000059	<0.0010	0.0012			<0.000010	0.0807
W8	7/29/2005	4.95	<0.00010	0.132			<0.010	<0.00010	0.000074	<0.0010	0.0019			<0.000010	0.033
W8	8/30/2005		<0.00050				<0.010	<0.00020	<0.00020	<0.030	<0.0050			<0.000020	0.0688
W8	9/28/2005		<0.00050				<0.010	<0.00020	<0.00020	<0.030	<0.0050			<0.000020	0.0509
W8	10/15/2005	6.54	<0.00010	0.171			<0.010	<0.00010	0.000093	<0.0010	<0.0010			<0.000010	0.116
W9	5/27/2005		<0.00050				<0.010	<0.00020	<0.00020	<0.030	<0.0050			<0.000020	0.0732
W9	6/30/2005	5.35	<0.00010	0.0957			<0.010	<0.00010	0.000024	<0.0010	<0.0010			<0.000010	0.0401
W9	7/29/2005	5.69	<0.00010	0.14			<0.010	<0.00010	0.00005	<0.0010	0.004			<0.000010	0.0478
W9	8/29/2005		<0.00050				<0.010	<0.00020	<0.00020	<0.030	<0.0050			<0.000020	0.0564
W9	9/28/2005		<0.00050				<0.010	<0.00020	<0.00020	<0.030	<0.0050			<0.000020	0.0642
W9	10/15/2005	6.13	<0.00010	0.0806			<0.010	<0.00010	0.000023	<0.0010	0.0017			<0.000010	0.0622
W10	5/27/2005		<0.00050				<0.010	<0.00020	<0.00020	<0.030	0.0085			<0.000020	0.151
W10	5/27/2005		<0.0005				<0.01	<0.0002	0.00079	<0.03	<0.005			<0.00002	
W10	6/30/2005	5.43	<0.00010	0.0772			<0.010	<0.00010	0.000028	<0.0010	0.0223			<0.000010	0.142
W10	7/29/2005	5.64	<0.00010	0.0921			<0.010	<0.00010	<0.000010	<0.0010	0.0074			<0.000030	2.59
W10	8/29/2005		<0.00050				<0.010	<0.00020	<0.00020	<0.030	0.0052			<0.000020	0.587
W10	9/28/2005		<0.00050				<0.010	<0.00020	<0.00020	<0.030	<0.0050			<0.000020	0.0818
W10	10/15/2005	5.82	<0.00010	0.147			<0.010	<0.00010	0.000013	<0.0010	0.0015			<0.000010	0.089
W6	5/27/2005		<0.00050				<0.010	<0.00020	<0.00020	<0.030	<0.0050			<0.000020	0.229
W6	6/30/2005	7.18	<0.00010	0.102			<0.010	<0.00010	0.000134	<0.0010	0.021			<0.000010	0.17
W6	7/29/2005	5.51	<0.00010	0.274			<0.010	<0.00010	0.0014	<0.0010	0.0025			<0.000020	0.598
W6	8/30/2005		<0.00050				<0.010	<0.00020	<0.00020	<0.030	<0.0050			<0.000020	0.0181
W6	9/28/2005		<0.00050				<0.010	<0.00020	<0.00020	<0.030	<0.0050			<0.000020	0.0321
W6	10/15/2005	6.93	<0.00010	0.108			<0.010	<0.00010	0.000122	<0.0010	0.0015			<0.000010	0.0251
W6	6/2/2006	4.84	<0.0001	0.0682			<0.01	<0.0001	0.000081	<0.001	0.003			0.000021	1.95
W6	8/25/2006													<0.0002	0.024
W6	9/6/2006	7.54	<0.001	0.136			0.0005	<0.00005	<0.0005	0.0016	0.002			<0.0001	0.035
W6	6/20/2007													<0.0001	0.042
W6	9/18/2007													<0.0001	0.023
W6	6/17/2008													<0.00001	0.14
W6	9/18/2008	6.29	<0.0001	0.098	<0.0001	<0.0001	0.001	<0.00001	<0.0004	0.00073	0.003	0.0002		0.00008	0.022
W6	4/14/2009	3.63	<0.0001	0.419	<0.0001	<0.0001	0.0032	<0.00001	0.004	0.00029	0.049	<0.0001		0.00021	1.3
W6	4/30/2009	2.3	<0.0001	0.099	<0.0001	<0.0001	0.0044	<0.00001	<0.0004	0.00047	0.006	0.0004		<0.00001	5.31
W6	7/28/2009														
W6	8/6/2009													<0.00001	0.085
W6	8/6/2009													<0.00001	0.07
W6	8/18/2009	4.06	<0.0001	0.584	<0.0001	<0.0004	0.0004	<0.00001	0.0008	0.0006	0.002	<0.0001			
W6	8/25/2009														
W6	10/7/2009														
W6	10/7/2009													<0.00001	0.171
W6	10/25/2009	4.2	<0.005	0.817			<0.005	<0.00005	0.0009	<0.005	<0.005	<0.0005		<0.00002	0.065
W6	4/19/2010	3.81	<0.0001	0.372	<0.0001	<0.0004	<0.01	<0.00001	0.0013	0.0006	0.005	<0.0001		0.00002	1.25
W6	5/16/2010	5.52	<0.005	0.199			<0.005	<0.00005	<0.0001	<0.005	<0.005	<0.0005		0.00002	0.041
W6	5/17/2010	2.22	<0.005	0.219			<0.005	<0.00005	0.0005	<0.005	<0.005	<0.0005		<0.00002	0.025
W6	5/19/2010	2.79	<0.005	0.245			<0.005	<0.00005	0.0005	<0.005	<0.005	<0.0005		<0.00002	0.03
W6	5/21/2010	2.14	<0.005	0.243			<0.01	<0.00005	0.0006	<0.005	<0.01	<0.002		<0.0001	0.162
W6	5/23/2010	2.58	<0.005	0.285			<0.01	<0.00005	0.0006	<0.005	<0.01	<0.002		<0.0001	0.027
W6	5/25/2010	2.6	<0.005	0.306			<0.01	<0.00005	0.0005	<0.005	<0.01	<0.002		<0.0001	0.061
W6	5/29/2010	<2	<0.005	0.221			<0.01	<0.00005	0.0005	<0.005	<0.01	<0.002		<0.0001	0.09
W6	6/2/2010	5.16	<0.005	0.623			<0.01	<0.00005	0.0028	<0.005	<0.01	<0.002		<0.0001	0.036
W6	6/4/2010	5.66	<0.005	0.099			<0.01	<0.00005	0.0003	<0.005	<0.01	<0.002		<0.0001	0.049
W6	6/6/2010	6.95	<0.005	0.115			<0.01	<0.00005	0.0004	<0.005	<0.01	<0.002		<0.0001	0.142
W6	6/14/2010	7.21	<0.005	0.099			<0.01	<0.00005	0.0003	<0.005	<0.01	<0.002		<0.0001	0.274
W6	6/16/2010	6.57	<0.005	0.117			<0.01	<0.00005	0.0003	<0.005	<0.01	<0.002		<0.0001	0.274
W6	6/18/2010	7.25	<0.005	0.111			<0.01	<0.00005	0.0003	<0.005	<0.01	<0.002		<0.0001	0.154
W6	6/20/2010	6.72	<0.005	0.109			<0.01	<0.00005	0.0003	<0.005	<0.01	<0.002		<0.0001	0.138
W6	6/22/2010	6.78	<0.005	0.109			<0.01	<0.00005	0.0003	<0.005	<0.01	<0.002		<0.0001	0.208
W6	6/24/2010	6.84	<0.005	0.109			<0.01	<0.00005	0.0003	<0.005	<0.01	<0.002		<0.0001	0.118
W6	6/26/2010	6.94	<0.005	0.118			<0.01	<0.00005	0.0003	<0.005	<0.01	<0.002		<0.0001	0.143
W6	6/26/2010														
W6	6/28/2010	7.64	<0.005	0.115			<0.01	<0.00005	0.0002	<0.005	<0.01	<0.002		<0.0001	0.612
W6	7/5/2010	6.24	<0.005	0.112			<0.01	<0.00005	0.0002	<0.005	<0.01	<0.002		<0.0001	1.41
W6	8/5/2010	7.84	<0.005	0.13			<0.01	<0.00005	0.0002	<0.005	<0.01	<0.002		<0.0001	0.157
W6	9/9/2010	7.77	<0.005	0.13			<0.01	<0.00005	0.0002	<0.005	<0.01	<0.002		<0.0001	0.655
W6	10/8/2010	6.98	<0.005	0.125			<0.005	<0.00005	0.0002	<0.005	<0.005	<0.0005		<0.00002	0.429
W6	8/19/2011	5.54	<0.005	0.617			<0.005	<0.00005	0.0025	<0.005	<0.005	<0.0005		<0.00002	0.033
W6	7/17/2012	7.25	<0.0050	0.131			<0.0050	<0.000050	0.00034	<0.0050	<0.0050	<0.00050		0.000172	20.8
W6	7/30/2012	6.95	<0.0050	0.135			<0.0050	<0.000050	0.00039	<0.0050	<0.0050	<0.00050		0.000122	15.1
W6	8/9/2012	8.13	<0.0050	0.151			<0.0050	<0.000050	0.00038	<0.0050	<0.0050	<0.00050		0.000054	8.03
W6	8/25/2012	6.56	<0.0050	0.143			<0.0050	<0.000050	0.0004	<0.0050	<0.0050	<0.00050		0.000021	5.23
W6	9/13/2012	6.42	<0.0050	0.11			<0.0050	<0.000050	0.00013	<0.0050	<0.0050	<0.00050		<0.000020	0.271
W6	10/13/2012													<0.000020	0.241
W6	11/12/2012	8.72	<0.0050	0.189			<0.0050	<0.000050	0.00032	<0.0050	<0.0050	<0.00050		<0.000020	0.0243
W6	5/14/2013	2.67	<0.0050	0.0363			<0.0050	<0.000050	<0.00010	<0.0050	<0.0050	<0.00050		<0.000020	0.65
W6	6/22/2013	6.57	<0.0050	0.115			<0.0050	<0.000050	0.00028	<0.0050	<0.0050	<0.00050		<0.000020	4.63
W6	7/12/2013	7.05	<0.0050	0.112			<0.0050	<0.000050	0.00017	<0.0050	<0.0050	<0.00050		<0.000020	0.506
W6	8/22/2013	7.24	<0.0050	0.132			<0.0050	<0.000050	0.00017	<0.0050	<0.0050	<0.00050		<0.000020	0.0846
W6	9/6/2013	7.48	<0.0050	0.132			<0.0050	<0.000050	0.00016	<0.0050	<0.0050	<0.00050		<0.000020	0.165
W6	10/7/2013	7.39	<0.0050	0.126			<0.0050	<0.000050	0.00018	<0.0050	<0.0050	<0.00050		0.000045	2.18
W6	11/16/2013	8.28	<0.0050	0.144			<0.0050	<0.000050	0.00019	<0.0050	<0.0050	<0.00050		<0.000020	0.283
W6	12/7/2013	9.38	<0.0050	0.2			<0.0050	<0.000050	0.00039	<0.0050	<0.0050	<0.00050		0.000049	0.0734
W7	5/27/2005		<0.00050				<0.010	<0.00020	0.00036	<0.030	<0.0050			<0.000020	0.489
W7	6/30/2005	5.72	<0.00010	0.284			<0.010	<0.00010	0.000759	<0.0010	0.0064			<0.000010	0.153
W7	7/29/2005	5.47	<0.00010	0.277			<0.010	<0.00010	0.000994	<0.0010	0.002			<0.000020	0.506
W7	8/30/2005		<0.00050				<0.010	<0.00020	0.00055	<0.030	<0.0050			<0.000020	0.0436
W7	9/28/2005		<0.00050				<0.010	<0.00020	0.00038	<0.030	<0.0050			<0.000020	1.65
W7	10/15/2005	6.44	<0.00010	0.23			<0.010	<0.00010	0.000515	<0.0010	<0.0010			<0.000010	0.0738
W7	6/2/2006	4.19	<0.0001	0.11			<0.01	<0.0001	0.000216	0.0017	<0.001			0.000069	11.4
W7	6/8/2006													<0.00001	0.648

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Dissolved Si (mg/L)	Dissolved Sn (mg/L)	Dissolved Sr (mg/L)	Dissolved Te (mg/L)	Dissolved Th (mg/L)	Dissolved Ti (mg/L)	Dissolved Tl (mg/L)	Dissolved U (mg/L)	Dissolved V (mg/L)	Dissolved Zn (mg/L)	Dissolved Zr (mg/L)	P-E (mg/L)	TT Ag (mg/L)	TT Al (mg/L)
W7	6/15/2006														
W7	6/15/2006	5.35	<0.001	0.236			0.0007	<0.00005	0.0007	0.0011	<0.001			<0.0001	1.48
W7	6/23/2006	5.5	<0.001	0.321			0.0006	<0.00005	0.001	0.001	0.002			0.0003	5.09
W7	6/28/2006	6.48	<0.001	0.284			0.0117	<0.00005	0.0009	0.0029	0.005			<0.0001	1.69
W7	7/7/2006	6.8	<0.001	0.322			0.0007	<0.00005	0.0005	0.0013	0.001			<0.0001	0.096
W7	7/12/2006	5.36	<0.001	0.322			0.0005	<0.00005	0.0007	0.0018	<0.001			<0.0001	0.078
W7	7/20/2006	6.34	<0.001	0.284			<0.0005	<0.00005	0.001	0.0016	<0.001			<0.0001	0.058
W7	7/26/2006														
W7	7/26/2006	5.35	<0.001	0.274			<0.0005	<0.00005	0.001	0.0007	<0.001			<0.0001	0.099
W7	8/2/2006	5.87	<0.001	0.247			0.0005	<0.00005	0.001	0.0013	<0.001			0.0002	0.06
W7	8/10/2006	5.9	<0.001	0.343			0.0006	<0.00005	0.001	0.0015	0.001			<0.0001	0.034
W7	8/25/2006														
W7	8/25/2006	6.32	<0.001	0.344			0.0005	<0.00005	0.0011	0.0017	0.003			<0.0002	0.048
W7	8/30/2006	5.84	<0.001	0.333			0.0008	<0.00005	0.001	0.0008	0.005			<0.0001	0.054
W7	9/6/2006	6.02	<0.001	0.355			0.0005	<0.00005	0.001	0.0018	0.002			<0.0001	0.049
W7	9/13/2006	5.56	<0.001	0.321			0.0006	<0.00005	0.001	0.0014	0.001			<0.0001	0.025
W7	9/20/2006	5.74	<0.001	0.319			<0.0005	<0.00005	0.001	0.0015	<0.001			<0.0001	0.074
W7	9/28/2006	5.97	<0.001	0.34			<0.0005	<0.00005	0.001	0.0015	0.002			<0.0002	0.042
W7	10/4/2006	5.93	<0.001	0.326			0.0007	<0.00005	0.001	0.0011	0.001			<0.0002	0.082
W7	10/12/2006	6.08	<0.001	0.344			0.0006	<0.00005	0.0011	0.0009	0.002			<0.0001	0.264
W7	6/5/2007													<0.0001	1.56
W7	6/5/2007													<0.0002	1.77
W7	6/5/2007													<0.0001	1.85
W7	6/20/2007	5.02	<0.001	0.307			0.0009	<0.00005	0.0012	0.0016	0.004			<0.0001	0.172
W7	7/18/2007	6.1	<0.001	0.34			0.0006	<0.00005	0.0012	0.0024	0.006			<0.0001	0.231
W7	7/18/2007													<0.0001	0.216
W7	7/18/2007													<0.0001	0.211
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007													<0.0001	0.205
W7	8/24/2007													<0.0001	0.098
W7	8/24/2007													<0.0001	0.092
W7	8/24/2007													<0.0001	0.092
W7	9/18/2007	6.17	<0.001	0.31			0.0006	<0.00005	0.0006	0.0044	0.002			<0.0001	0.33
W7	10/30/2007													<0.0001	0.121
W7	10/30/2007													<0.0001	0.071
W7	10/30/2007													<0.0001	0.076
W7	10/30/2007														
W7	4/22/2008	3.97	<0.004	0.497	<0.0001	<0.0001	0.0011	<0.00001	0.0021	0.00059	0.001	<0.0001		<0.0001	<0.02
W7	4/22/2008	4.03	<0.004	0.51	<0.0001	<0.0001	0.0009	<0.00001	0.0024	0.00077	0.003	0.0001		<0.0001	0.02
W7	4/22/2008	3.95	<0.004	0.491	<0.0001	<0.0001	0.0014	<0.00001	0.0022	0.0003	0.001	<0.0001		<0.0001	0.02
W7	4/22/2008														
W7	6/3/2008													<0.0001	0.4
W7	6/3/2008													<0.0001	0.38
W7	6/3/2008													<0.0001	0.42
W7	6/3/2008	2.27	<0.0001	0.226	<0.0001	<0.0001	0.0005	<0.00001	0.001	0.00062	0.002	0.0004			
W7	6/17/2008		<0.0001	0.289	0.0002	<0.0001	0.0015	<0.00001	0.0014	0.00091	0.002	0.0003		<0.00001	0.21
W7	8/6/2008													0.00001	0.817
W7	8/6/2008													0.00001	1.27
W7	8/6/2008													0.00001	1.36
W7	8/6/2008	5.81	<0.001	0.336			<0.0005	<0.00005	0.0011	0.001	0.006				
W7	9/18/2008	5.8	<0.0001	0.173	<0.0001	<0.0001	0.0055	<0.00001	<0.0004	0.00094	0.003	0.0004		0.00006	0.469
W7	10/28/2008													<0.00001	0.084
W7	10/28/2008													<0.00001	0.038
W7	10/28/2008													<0.00001	0.031
W7	10/28/2008	6.6	<0.0001	0.276	<0.0001	<0.0001	0.0011	<0.00001	0.0007	0.00067	0.001	0.0002			
W7	4/30/2009	1.48	<0.0001	0.029	<0.0001	<0.0001	0.0015	<0.00001	<0.0004	0.00024	0.009	0.0002		<0.00001	0.138
W7	6/8/2009	4.13	<0.0001	0.24	<0.0001	<0.0001	0.0011	<0.00001	0.0008	0.0008	0.002	0.0003		0.00002	0.595
W7	7/28/2009														
W7	8/6/2009													0.00002	0.071
W7	8/24/2009														
W7	8/25/2009														
W7	8/31/2009	6.04	<0.0001	0.256	<0.0001	<0.0004	0.0004	<0.00001	0.0007	0.0007	0.001	0.0002			
W7	10/7/2009														
W7	10/7/2009													<0.00001	0.189
W7	10/25/2009	6.3	<0.005	0.334			<0.005	<0.00005	0.0012	<0.005	<0.005	<0.0005		<0.00002	0.052
W7	10/25/2009														
W7	4/19/2010	4.15	<0.0001	0.212	<0.0001	<0.0004	<0.01	<0.00001	0.0005	0.0005	0.005	<0.0001		<0.00001	0.033
W7	4/27/2010	2.94	<0.005	0.071			<0.005	<0.00005	0.0001	<0.005	<0.005	<0.0005		<0.00002	0.113
W7	4/29/2010	2.89	<0.005	0.076			<0.005	<0.00005	0.0001	<0.005	<0.005	<0.0005		<0.00002	0.11
W7	5/1/2010	3.27	<0.005	0.088			<0.005	<0.00005	0.0002	<0.005	<0.005	<0.0005		<0.00002	0.502
W7	5/3/2010	3.63	<0.005	0.097			<0.005	<0.00005	0.0002	<0.005	<0.005	<0.0005		0.00003	0.406
W7	5/5/2010	3.84	<0.005	0.116			<0.005	<0.00005	0.0003	<0.005	<0.005	<0.0005		<0.00002	0.394
W7	5/7/2010	3.77	<0.005	0.129			<0.005	<0.00005	0.0003	<0.005	<0.005	<0.0005		<0.00002	0.407
W7	5/9/2010	4.61	<0.005	0.192			<0.005	<0.00005	0.0006	<0.005	<0.005	<0.0005		<0.00002	0.183
W7	5/10/2010	4.62	<0.005	0.195			<0.005	<0.00005	0.0006	<0.005	<0.005	<0.0005		<0.00002	0.125
W7	5/12/2010														
W7	5/12/2010	5	<0.005	0.201			<0.005	<0.00005	0.0006	<0.005	<0.005	<0.0005		<0.00002	0.123
W7	5/14/2010	4.35	<0.005	0.178			<0.005	<0.00005	0.0005	<0.005	<0.005	<0.0005		<0.00002	0.286
W7	5/16/2010	4.68	<0.005	0.207			<0.005	<0.00005	0.0007	<0.005	<0.005	<0.0005		<0.00002	0.207
W7	5/17/2010	4.22	<0.005	0.191			<0.005	<0.00005	0.0006	<0.005	<0.005	<0.0005		<0.00002	0.219
W7	5/19/2010	5.24	<0.005	0.21			<0.005	<0.00005	0.0008	<0.005	<0.005	<0.0005		<0.00002	0.105
W7	5/21/2010	4.98	<0.005	0.22			<0.01	<0.00005	0.0008	<0.005	<0.01	<0.002		<0.0001	0.457
W7	5/23/2010	5.24	<0.005	0.235			<0.01	<0.00005	0.0009	<0.005	<0.01	<0.002		<0.0001	2.14
W7	5/25/2010	6.05	<0.005	0.306			<0.01	<0.00005	0.0011	<0.005	<0.01	<0.002		<0.0001	4.2
W7	5/27/2010	5.4	<0.005	0.277			<0.01	<0.00005	0.0011	<0.005	<0.01	<0.002		<0.0001	10.3
W7	5/29/2010	5.26	<0.005	0.289			<0.01	<0.00005	0.0014	<0.005	<0.01	<0.002		<0.0001	3.93
W7	5/31/2010		<0.005	0.286			<0.01	<0.00005	0.0014	<0.005	<0.01	<0.002		<0.0001	4.54
W7	6/2/2010	5.16	<0.005	0.293			<0.01	<0.00005	0.0015	<0.005	<0.01	<0.002		<0.0001	2.91
W7	6/4/2010	4.96	<0.005	0.275			<0.01	<0.00005	0.0014	<0.005	<0.01	<0.002		<0.0001	1.64
W7	6/6/2010	5.46	<0.005	0.313			<0.01	<0.00005	0.0015	<0.005	<0.01	<0.002		<0.0001	2.05
W7	6/14/2010	5.87	<0.005	0.314			<0.01	<0.00005	0.0015	<0.005	<0.01	<0.002		<0.0001	1.1
W7	6/16/2010	5.32	<0.005	0.338			<0.01	<0.00005	0.0015	<0.005	<0.01	<0.002		<0.0001	0.43
W7	6/18/2010	5.82	<0.005	0.318			<0.01	<0.00005	0.0015	<0.005	<0.01	<0.002		<0.0001	0.504
W7	6/20/2010	5.69	<0.005	0.3			<0.01	<0.00005	0.0012	<0.005	<0.01	<0.002		<0.0001	1.17
W7	6/22/2010	5.86	<0.005	0.249			<0.01	<0.00005	0.0008	<0.005	<0.01	<0.002		<0.0001	5.52
W7	6/24/2010	5.79	<0.005	0.259			<0.01	<0.00005	0.0008	<0.005	<0.01	<0.002		<0.0001	1.21
W7	6/26/2010	6.05	<0.005	0.305			<0.01	<0.00005	0.001	<0.005	<0.01	<0.002		<0.0001	0.503
W7	6/28/2010	6.64	<0.005	0.314			<0.01	<0.00005	0.001	<0.005	<0.01	<0.002		<0.0001	0.298
W7	7/5/2010	6.08	<0.005	0.255			<0.01	<0.00005	0.0006	<0.005	<0.01	<0.002		<0.0001	1.59
W7	8/5/2010	6.47	<0.005	0.303			<0.01	<0.00005	0.0007	<0.005	<0.01	<0.002		<0.0001	0.22
W7	9/12/2010													<0.0001	0.256
W7	9/12/2010	7.17	<0.005	0.253			<0.01	<0.00005	0.0005	<0.005	<0.01	<0.002		<0.0001	0.273
W7	10/8/2010	6.54	<0.005	0.277			<0.005	<0.00005	0.0007	<0.005	<0.005	<0.0005		<0.00002	0.064
W7	4/27/2011	3.09	<0.005	0.084			<0.005	<0.00005	0.0001	<0.005	0.005	<0.0005		<0.00002	0.101
W7	5/2/2011	2.52	<0.005	0.062			<0.005	<0.00005	<0.0001	<0.005	0.01	<0.0005		<0.00002	0.25
W7	6/1/2011	5.26	<0.005	0.25			<0.005	<0.00005	0.0007	<0.005	<0.005	<0.0005		<0.00002	0.454
W7	6/16/2011	6.52	<0.005	0.213			<0.005	<0.00005	0.0005	<0.005	<0.005	<0.0005		<0.00002	2.48
W7	7/1/2011	6.61	<0.005	0.154			<0.005	<0.00005	0.0003	<0.005	<0.005	<0.0005		0.00004	3.4

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Dissolved Si (mg/L)	Dissolved Sn (mg/L)	Dissolved Sr (mg/L)	Dissolved Te (mg/L)	Dissolved Th (mg/L)	Dissolved Ti (mg/L)	Dissolved Tl (mg/L)	Dissolved U (mg/L)	Dissolved V (mg/L)	Dissolved Zn (mg/L)	Dissolved Zr (mg/L)	P-E (mg/L)	TT Ag (mg/L)	TT Al (mg/L)
W7	8/2/2011	5.78	<0.005	0.12			<0.005	<0.00005	0.0002	<0.005	<0.005	<0.0005		0.00008	9.23
W7	8/16/2011	6.96	<0.005	0.21			<0.005	<0.00005	0.0004	<0.005	<0.005	<0.0005		<0.00002	0.403
W7	8/19/2011	6.9	<0.005	0.197			<0.005	<0.00005	0.0003	<0.005	<0.005	<0.0005		<0.00002	0.444
W7	9/12/2011	7.71	<0.005	0.266			<0.005	<0.00005	0.0005	<0.005	<0.005	<0.0005		<0.00002	0.08
W7	10/6/2011	6.17	<0.005	0.266			<0.005	<0.00005	0.0006	<0.005	<0.005	<0.0005		<0.00002	0.065
W7	11/22/2011	8.07	<0.0050	0.366			<0.0050	<0.000050	0.00106	<0.0050	<0.0050	<0.00050		<0.00002	0.407
W7	1/24/2012	7.24	<0.0050	0.456			<0.0050	<0.000050	0.00163	<0.0050	<0.0050	<0.00050		<0.00002	0.01
W7	2/25/2012	6.43	<0.0050	0.448			<0.0050	<0.000050	0.0015	<0.0050	0.0083	<0.00050		<0.00002	0.018
W7	3/5/2012	5.99	<0.0050	0.468			<0.0050	<0.000050	0.00145	<0.0050	<0.0050	<0.00050		<0.00002	0.043
W7	4/24/2012	2.94	<0.0050	0.0774			<0.0050	<0.000050	0.00014	<0.0050	<0.0050	<0.00050		<0.000020	0.766
W7	5/6/2012	3.66	<0.0050	0.106			<0.0050	<0.000050	<0.00010	<0.0050	<0.0050	<0.00050		<0.000020	0.11
W7	5/6/2012	3.7	<0.0050	0.105			<0.0050	<0.000050	<0.00010	<0.0050	<0.0050	<0.00050		<0.000020	0.131
W7	6/19/2012	6.21	<0.0050	0.191			<0.0050	<0.000050	0.00043	<0.0050	<0.0050	<0.00050		0.00003	5.38
W7	7/17/2012	6.79	<0.0050	0.306			<0.0050	<0.000050	0.00074	<0.0050	<0.0050	<0.00050		<0.000020	0.93
W7	7/30/2012	6.86	<0.0050	0.329			<0.0050	<0.000050	0.00091	<0.0050	<0.0050	<0.00050		<0.000020	0.2
W7	8/8/2012	7	<0.0050	0.357			<0.0050	<0.000050	0.00096	<0.0050	<0.0050	<0.00050		<0.000020	0.0828
W7	8/23/2012	7.06	<0.0050	0.35			<0.0050	<0.000050	0.00097	<0.0050	<0.0050	<0.00050		<0.000020	0.0342
W7	9/13/2012	6.7	<0.0050	0.247			<0.0050	<0.000050	0.00052	<0.0050	<0.0050	<0.00050		<0.000020	0.139
W7	10/16/2012	6.8	<0.0050	0.303			<0.0050	<0.000050	0.00077	<0.0050	<0.0050	<0.00050		<0.000020	0.127
W7	11/2/2012	6.94	<0.0050	0.407			<0.0050	<0.000050	0.00135	<0.0050	<0.0050	<0.00050		<0.000020	0.0576
W7	12/29/2012	9.63	<0.0050	0.535			<0.0050	<0.000050	0.00133	<0.0050	<0.0050	<0.00050		<0.000020	0.0127
W7	4/30/2013	6.83	<0.0050	0.607			<0.0050	<0.000050	0.00213	<0.0050	<0.0050	<0.00050		<0.000020	0.0111
W7	5/14/2013	2.15	<0.0050	0.063			<0.0050	<0.000050	<0.00010	<0.0050	<0.0050	<0.00050		<0.000020	1.34
W7	6/19/2013	5.93	<0.0050	0.276			<0.0050	<0.000050	0.00097	<0.0050	<0.0050	<0.00050		<0.000020	0.256
W7	6/19/2013	5.92	<0.0050	0.277			<0.0050	<0.000050	0.00098	<0.0050	<0.0050	<0.00050		<0.000020	0.213
W7	7/13/2013	7.01	<0.0050	0.327			<0.0050	<0.000050	0.00105	<0.0050	<0.0050	<0.00050		0.000024	0.127
W7	8/21/2013	6.79	<0.0050	0.364			<0.0050	<0.000050	0.00107	<0.0050	<0.0050	<0.00050		<0.000020	0.137
W7	9/6/2013	7.12	<0.0050	0.339			<0.0050	<0.000050	0.00088	<0.0050	<0.0050	<0.00050		<0.000020	0.37
W7	10/21/2013	7.18	<0.0050	0.353			<0.0050	<0.000050	0.00093	<0.0050	<0.0050	<0.00050		<0.000020	0.0931
W7	11/16/2013	6.47	<0.0050	0.418			<0.0050	<0.000050	0.0015	<0.0050	<0.0050	<0.00050		<0.000020	0.0389
W7	12/7/2013	7.02	<0.0050	0.439			<0.0050	<0.000050	0.00126	<0.0050	<0.0050	<0.00050		0.000046	0.0098
C4	5/6/2012	4.02	<0.0050	0.0653			<0.0050	<0.000050	0.00015	<0.0050	<0.0050	0.00054		0.00021	29.7
C4	8/1/2012	8.03	<0.0050	0.206			<0.0050	<0.000050	0.00065	<0.0050	<0.0050	0.00061		0.000056	7.06
C4	8/8/2012	7.91	<0.0050	0.225			<0.0050	<0.000050	0.00075	<0.0050	<0.0050	0.00053		0.000051	7.01
C4	8/25/2012	7.53	<0.0050	0.231			<0.0050	<0.000050	0.00076	<0.0050	<0.0050	0.00058		0.00002	4.45
C4	6/22/2013	6.82	<0.0050	0.184			<0.0050	<0.000050	0.00115	<0.0050	<0.0050	0.00052		0.000074	20.4
C4	7/19/2013	7.14	<0.0050	0.193			<0.0050	<0.000050	0.001	<0.0050	<0.0050	<0.00050		0.000087	13.5
C4	8/22/2013	7.18	<0.0050	0.209			<0.0050	<0.000050	0.00095	<0.0050	<0.0050	0.00058		0.000063	13.8
C4	9/6/2013	7.73	<0.0050	0.214			<0.0050	<0.000050	0.00075	<0.0050	<0.0050	0.00055		0.000021	5.43
C4	10/18/2013	7.06	<0.0050	0.217			<0.0050	<0.000050	0.00068	<0.0050	<0.0050	<0.00050		0.000052	0.349
C4	10/18/2013	7.12	<0.0050	0.218			<0.0050	<0.000050	0.0007	<0.0050	<0.0050	<0.00050		0.000124	0.358
C4	11/17/2013	8.74	<0.0050	0.405			<0.0050	<0.000050	0.00063	<0.0050	<0.0050	0.00111		<0.000020	0.0856
C10	5/12/2012	3.83	<0.0050	0.0919			<0.0050	<0.000050	0.00057	<0.0050	<0.0050	<0.00050		0.000337	39.6
C10	7/30/2012	6.28	<0.0050	0.189			<0.0050	<0.000050	0.00243	<0.0050	<0.0050	<0.00050		0.0002	21.3
C10	7/30/2012														
C10	8/10/2012	6.7	<0.0050	0.2			<0.0050	<0.000050	0.00182	<0.0050	<0.0050	<0.00050		0.000187	21.1
C10	8/27/2012	6.39	<0.0050	0.214			<0.0050	<0.000050	0.00149	<0.0050	<0.0050	<0.00050		0.000217	24.8
C10	6/22/2013	6.15	<0.0050	0.195			<0.0050	<0.000050	0.0011	<0.0050	0.0104	<0.00050		<0.000020	1.31
C10	7/23/2013	6.42	<0.0050	0.177			<0.0050	<0.000050	0.00075	<0.0050	<0.0050	<0.00050		0.000021	4.86
C10	8/22/2013	7.14	<0.0050	0.242			<0.0050	<0.000050	0.00117	<0.0050	<0.0050	<0.00050		<0.000020	0.947
C10	9/8/2013	7.12	<0.0050	0.255			<0.0050	<0.000050	0.00094	<0.0050	<0.0050	<0.00050		<0.000020	0.509
C10	9/8/2013	7.01	<0.0050	0.254			<0.0050	<0.000050	0.0009	<0.0050	<0.0050	<0.00050		<0.000020	0.483
C10	10/18/2013	5.99	<0.0050	0.201			<0.0050	<0.000050	0.0007	<0.0050	<0.0050	<0.00050		0.000068	1.14

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	TT As (mg/L)	Ba (mg/L)	Total Be (mg/L)	Total Bi (mg/L)	Total B (mg/L)	Total Ca (mg/L)	Total Cd (mg/L)	Total Co (mg/L)	Total Cr (mg/L)	Total Cu (mg/L)	Total Fe (mg/L)	Total Hg (mg/L)	Total K (mg/L)	Total La (mg/L)
W1	5/27/2005	0.0009	0.062	<0.0010		<0.10	25.9	<0.000050	0.00058	0.002	0.0059	1.05	<0.000020	<2.0	
W1	6/30/2005	0.00047	0.0545	<0.00050	<0.00050	<0.010	36.7	<0.000050	<0.00010	<0.00050	0.00204	0.088		<2.0	
W1	7/28/2005	0.00045	0.0597	<0.00050	<0.00050	<0.010	38.5	<0.000050	<0.00010	<0.00050	0.00229	0.079		<2.0	
W1	8/29/2005	0.00062	0.059	<0.0010		<0.10	35	<0.000017	<0.00030	<0.0010	0.0026	0.387	<0.000020	<2.0	
W1	9/28/2005	0.00068	0.059	<0.0010		<0.10	35.5	<0.000017	<0.00030	<0.0010	0.0039	0.691	<0.000020	<2.0	
W1	10/15/2005	0.00062	0.054	<0.00050	<0.00050	<0.010	34.2	<0.000050	0.00026	0.00068	0.00458	0.526		<2.0	
W2	5/27/2005	0.00075	0.058	<0.0010		<0.10	24.8	<0.000050	0.00043	0.0012	0.0049	0.778	<0.000020	<2.0	
W2	5/27/2005	0.00104	0.053	<0.001		<0.1	19.9	<0.00005	0.00087	0.0027	0.0037	1.49			
W2	6/30/2005	0.00045	0.0569	<0.00050	<0.00050	<0.010	37.5	<0.000050	<0.00010	<0.00050	0.00207	0.052		<2.0	
W2	7/28/2005	0.00046	0.0576	<0.00050	<0.00050	<0.010	39.1	<0.000050	<0.00010	<0.00050	0.00245	0.079		<2.0	
W2	8/30/2005	0.00063	0.06	<0.0010		<0.10	35.4	<0.000017	<0.00030	<0.0010	0.0026	0.373	<0.000020	<2.0	
W2	9/28/2005	0.00069	0.057	<0.0010		<0.10	34.3	<0.000017	<0.00030	<0.0010	0.0039	0.674	<0.000020	<2.0	
W2	10/15/2005	0.00061	0.0528	<0.00050	<0.00050	<0.010	34.1	<0.000050	0.00018	0.00059	0.00404	0.436		<2.0	
W3	5/27/2005	<0.00050	0.042	<0.0010		<0.10	20.9	<0.000050	<0.00030	<0.0010	0.0078	0.32	<0.000020	<2.0	
W3	6/30/2005	0.00037	0.0518	<0.00050	<0.00050	<0.010	39.1	<0.000050	<0.00010	0.0005	0.00374	0.081		<2.0	
W3	7/29/2005	0.00081	0.0667	<0.00050	<0.00050	<0.010	37.3	<0.000050	0.00062	0.002	0.0149	1.32		<2.0	
W3	8/29/2005	0.00054	0.06	<0.0010		<0.10	30.8	<0.000017	<0.00030	<0.0010	0.0086	0.698	<0.000020	<2.0	
W3	9/28/2005	0.00055	0.055	<0.0010		<0.10	28.7	<0.000017	0.0003	<0.0010	0.01	0.794	<0.000020	<2.0	
W3	10/15/2005	0.00045	0.0489	<0.00050	<0.00050	<0.010	29.4	<0.000050	0.00023	<0.00050	0.00855	0.497		<2.0	
W8	5/27/2005	<0.00050	0.04	<0.0010		<0.10	16.9	<0.000050	<0.00030	<0.0010	0.0143	0.23	<0.000020	<2.0	
W8	6/30/2005	0.00048	0.0503	<0.00050	<0.00050	<0.010	24.3	<0.000050	0.00021	0.00073	0.0103	0.503		<2.0	
W8	7/29/2005	0.00047	0.0546	<0.00050	<0.00050	<0.010	30.9	<0.000050	0.00017	0.00056	0.00891	0.435		<2.0	
W8	8/30/2005	0.00052	0.055	<0.0010		<0.10	25.6	<0.000017	<0.00030	<0.0010	0.0094	0.722	<0.000020	<2.0	
W8	9/28/2005	<0.00050	0.049	<0.0010		<0.10	24.7	<0.000017	<0.00030	<0.0010	0.0088	0.706	<0.000020	<2.0	
W8	10/15/2005	0.00054	0.0504	<0.00050	<0.00050	<0.010	25.1	<0.000050	0.00037	<0.00050	0.0104	0.887		<2.0	
W9	5/27/2005	<0.00050	0.042	<0.0010		<0.10	12.9	<0.000050	<0.00030	<0.0010	0.0035	0.277	0.000023	<2.0	
W9	6/30/2005	0.00058	0.0532	<0.00050	<0.00050	<0.010	18.9	<0.000050	0.00019	<0.00050	0.0021	0.799		<2.0	
W9	7/29/2005	0.00061	0.058	<0.00050	<0.00050	<0.010	25.1	<0.000050	0.0002	0.0006	0.00345	0.894		<2.0	
W9	8/29/2005	0.00064	0.054	<0.0010		<0.10	18.3	<0.000017	0.00031	<0.0010	0.002	1.01	<0.000020	<2.0	
W9	9/28/2005	0.00051	0.048	<0.0010		<0.10	17.2	<0.000017	<0.00030	<0.0010	0.0024	0.783	<0.000020	<2.0	
W9	10/15/2005	0.00042	0.0424	<0.00050	<0.00050	<0.010	15.6	<0.000050	0.00016	0.00051	0.00214	0.543		<2.0	
W10	5/27/2005	<0.00050	0.053	<0.0010		<0.10	12.9	<0.000050	<0.00030	<0.0010	0.02	0.113	<0.000020	<2.0	
W10	5/27/2005	<0.0005		<0.001		<0.1			0.00088	<0.001			<0.00002	<2.0	
W10	6/30/2005	0.00076	0.0459	<0.00050	<0.00050	<0.010	13.1	<0.000050	0.00063	0.00069	0.00534	1.01		<2.0	
W10	7/29/2005	0.00095	0.0868	<0.00050	<0.00050	<0.010	15.5	0.000097	0.00241	0.00499	0.0138	2.84		<2.0	
W10	8/29/2005	<0.00050	0.059	<0.0010		<0.10	18.5	0.000038	0.00053	0.0012	0.0056	0.681	<0.000020	<2.0	
W10	9/28/2005	<0.00050	0.057	<0.0010		<0.10	25.4	<0.000017	<0.00030	<0.0010	0.0035	0.097	<0.000020	<2.0	
W10	10/15/2005	0.0003	0.051	<0.00050	<0.00050	<0.010	24	<0.000050	0.00011	<0.00050	0.0045	0.121		<2.0	
W6	5/27/2005	0.00052	0.038	<0.0010		<0.10	17.4	<0.000050	<0.00030	<0.0010	0.0021	0.363	<0.000020	<2.0	
W6	6/30/2005	0.00045	0.0462	<0.00050	<0.00050	<0.010	24.7	<0.000050	0.00017	0.00057	0.00148	0.266		<2.0	
W6	7/29/2005	0.00063	0.0683	<0.00050	<0.00050	<0.010	41.1	<0.000050	0.00049	0.00145	0.0118	1.08		<2.0	
W6	8/30/2005	<0.00050	0.053	<0.0010		<0.10	28.6	<0.000017	<0.00030	<0.0010	0.0012	0.088	<0.000020	<2.0	
W6	9/28/2005	<0.00050	0.048	<0.0010		<0.10	28.2	<0.000017	<0.00030	<0.0010	0.0012	0.148	<0.000020	<2.0	
W6	10/15/2005	0.00043	0.0424	<0.00050	<0.00050	<0.010	25.1	<0.000050	0.00011	<0.00050	0.00107	0.116		<2.0	
W6	6/2/2006	0.00131	0.0679	<0.0005	<0.0005	<0.01	18.9	<0.00005	0.00127	0.00384	0.00531	2.74		<2	
W6	8/25/2006	0.0006	0.056	<0.0002	<0.001	0.004	31.5	<0.00002	<0.0002	<0.001	<0.002	<0.2		<0.8	
W6	9/6/2006	0.0005	0.052	<0.0001	<0.0005	0.003	29	<0.00001	0.0001	0.0005	0.001	0.1		0.5	
W6	6/20/2007	0.0004	0.042	<0.0001	<0.0005	0.004	23.2	<0.00001	0.0001	<0.0005	0.002	0.1	<0.0001	0.4	
W6	9/18/2007	0.0004	0.046	<0.0001	<0.0005	0.002	25.8	<0.00001	0.0001	<0.0005	0.001	<0.1	<0.0001	0.4	
W6	6/17/2008	0.0004	0.042	<0.00004		<0.005	23.8	<0.00008	0.00018	0.0011	0.001	0.35		0.46	
W6	9/18/2008	<0.0002	0.039	<0.00004	<0.0001	<0.005	24.6	<0.00001	0.00008	0.001	0.001	0.09	<0.00001	0.4	
W6	4/14/2009	0.0006	0.08	0.00006	<0.0001	0.017	36	0.00024	0.00079	0.0029	0.454	2.19	0.00002	9.2	
W6	4/30/2009	0.0023	0.111	0.00014	<0.0001	0.006	18.6	0.00008	0.00275	0.0096	0.068	8	0.00002	2.6	
W6	7/28/2009														
W6	8/6/2009	0.0008	0.098	<0.00004	<0.001	0.025	49.4	0.00002	0.00016	0.0008	0.011	0.053		3.4	
W6	8/6/2009	0.0008	0.09	<0.00004	<0.001	0.018	49.7	0.00003	0.00015	0.0008	0.011	0.055		3.4	
W6	8/18/2009														
W6	8/25/2009														
W6	10/7/2009														
W6	10/7/2009	0.0002	0.132	<0.00004	<0.001	0.039	58	<0.00001	0.00025	0.001	0.004	0.059	<0.00001	3.2	
W6	10/25/2009	0.0003	0.129	<0.0001	<0.001	<0.05	59.9	0.00002	<0.0005	<0.001	0.0066	0.076	<0.00002	3.41	
W6	4/19/2010	0.0006	0.068	<0.00004	<0.001	0.013	35.2	0.00003	0.00075	0.0012	0.051	1.73	<0.00001	2.7	
W6	5/16/2010	0.0004	0.024	<0.0001	<0.001	<0.05	25.1	<0.00001	<0.0005	<0.001	0.0051	0.08	<0.0002	1.34	
W6	5/17/2010	0.0002	0.028	<0.0001	<0.001	<0.05	19.6	0.00002	<0.0005	<0.001	0.0064	0.073	<0.0002	0.89	
W6	5/19/2010	<0.001	0.045	<0.0005	<0.001	<0.05	27.2	0.00003	<0.0005	0.001	0.007	0.069	<0.00002	1.32	
W6	5/21/2010	<0.0004	0.046	<0.0002	<0.001	<0.1	25	<0.0001	<0.0005	<0.002	0.033	0.263	<0.0002	1	
W6	5/23/2010	<0.0004	0.047	<0.0002	<0.001	<0.1	27	<0.0001	<0.0005	<0.002	0.007	0.083	<0.0002	1	
W6	5/25/2010	<0.001	0.049	<0.0005	<0.001	<0.1	32	<0.0001	<0.0005	<0.002	0.028	0.133	<0.00002	1	
W6	5/29/2010	<0.0004	0.04	<0.0002	<0.001	<0.1	21	<0.0001	<0.0005	<0.002	0.022	0.21	<0.00002	<1	
W6	6/2/2010	<0.0004	0.077	<0.0002	<0.001	<0.1	61	<0.0001	<0.0005	<0.002	0.002	0.081	<0.0002	2	
W6	6/4/2010	0.0004	0.047	<0.0002	<0.001	<0.1	26	<0.0001	<0.0005	<0.002	0.002	0.12	<0.0002	<1	
W6	6/6/2010	0.0005	0.053	<0.0002	<0.001	<0.1	27	<0.0001	<0.0005	<0.002	0.001	0.218		<1	
W6	6/14/2010	0.0005	0.054	<0.0002	<0.001	<0.1	26	<0.0001	<0.0005	<0.002	0.004	0.378	<0.0002	<1	
W6	6/16/2010	0.0005	0.05	<0.0002	<0.001	<0.1	25	<0.0001	<0.0005	<0.002	0.005	0.364	<0.0002	<1	
W6	6/18/2010	0.0005	0.048	<0.0002	<0.001	<0.1	24	<0.0001	<0.0005	<0.002	0.003	0.226	<0.0002	<1	
W6	6/20/2010	0.0004	0.049	<0.0002	<0.001	<0.1	25	<0.0001	<0.0005	<0.002	0.002	0.208	<0.0002	<1	
W6	6/22/2010	0.0006	0.051	<0.0002	<0.001	<0.1	27	<0.0001	<0.0005	<0.002	0.002	0.326	<0.0002	<1	
W6	6/24/2010	0.0004	0.04	<0.0002	<0.001	<0.1	21	<0.0001	<0.0005	<0.002	0.001	0.164	<0.0002	<1	
W6	6/26/2010	<0.0004	0.047	<0.0002	<0.001	<0.1	25	<0.0001	<0.0005	<0.002	0.002	0.215	<0.0002	<1	
W6	6/26/2010														
W6	6/28/2010	0.0006	0.056	<0.0002	<0.001	<0.1	26	<0.0001	<0.0005	<0.002	0.003	0.8	<0.0002	<1	
W6	7/5/2010	0.001	0.072	<0.0002	<0.001	<0.1	24	<0.0001	0.0009	0.003	0.005	1.93	<0.0002	<1	
W6	8/5/2010	0.0005	0.056	<0.0002	<0.001	<0.1	28	<0.0001	<0.0005	<0.002	0.003	0.294	<0.0002	<1	
W6	9/9/2010	0.0005	0.07	<0.0002	<0.001	<0.1	29	<0.0001	<0.0005	<0.002	0.002	0.949	<0.0002	<1	
W6	10/8/2010	0.0003	0.065	<0.0001	<0.001	<0.05	25.5	0.00002	<0.0005	0.001	0.0021	0.615	<0.00002	0.5	
W6	8/19/2011	0.0003	0.077	<0.0001	<0.001	<0.05	51.2	0.00001	<0.0005	<0.001	0.003	0.059	<0.00002	2.02	
W6	7/17/2012	0.00898	0.471	0.00066	<0.0010	<0.05	41.2	0.000468	0.014	0.0388	0.0402	29.8	<0.000010	2.03	
W6	7/30/2012	0.0066	0.348	0.00043	<0.0010	<0.05	38.1	0.000195	0.00922	0.0276	0.0286	21.6	0.000033	1.62	
W6	8/9/2012	0.00362	0.223	0.00031	<0.0010	<0.05	36.7	0.000109	0.00511	0.0137	0.0147	11.2	<0.000010	1.2	
W6	8/25/2012	0.00395	0.372	0.00044	<0.0010	<0.05	45	0.000219	0.00749	0.0085	0.0247	9.68	<0.000010	0.849	
W6	9/13/2012	0.00051	0.0533	<0.00010	<0.0010	<0.05	25.9	0.000011	<0.00050	<0.0010	0.00225	0.591	<0.000010	0.42	
W6	10/13/2012	0.00066	0.0631	<0.00010	<0.0010	<0.05	30.8	<0.000010	<0.00050	<0.0010	0.00189	0.517	<0.000010	0.594	
W6	11/12/2012	0.00042	0.0656	<0.00010	<0.0010	<0.05	43.3	<0.000010	<0.00050	<0.0010	0.00345	0.0646	<0.000010	0.507	
W6	5/14/2013	0.00068	0.0286	<0.00010	<0.0010	<0.05	9.4	0.000029	0.0006	0.0014	0.00697	1.06	<0.000010	1.86	
W6	6/22/2013	0.00264	0.127	0.00017	<0.0010	<0.05	30.4	0.000048	0.00283	0.0094	0.00985	6.84	<0.000010	1.45	
W6	7/12/2013	0.00066	0.0579	<0.00010	<0.0010	<0.05	26.7	<0.000010	<0.00050	0.0011	0.00171	0.812	<0.000010	0.55	
W6	8/22/2013	0.00064	0.0507	<0.00010	<0.0010	<0.05	29.6	<0.000010	<0.00050	<0.0010	0.00123	0.283	<0.000010	0.52	
W6	9/6/2013	0.00067	0.0563	<0.00010	<0.0010	<0.05	31.7	<0.000010	<0.00050	<0.0010	0.00119	0.404	<0.000010	0.605	
W6	10/7/2013	0.00138	0.0866	<0.00010	<0.0010	<0.05	29.2	0.000023	0.00119	0.0043	0.0071	3.02	<0.000010	0.789	
W6	11/16/2013	0.00063	0.0637	<0.00010	<0.0010	<0.05	33.4	<0.000010	<0.00050	<0.0010	0.00186	0.523	<0.000010	0.652	
W6	12/7/2013	0.00074	0.0801	<0.00010	<0.0010	<0.05	49.1	0.00002	<0.00050	<0.0010	0.00602	0.12	<0.000010	0.661	
W7	5/27/2005	0.00063	0.054	<0.0010		<0.10	19.3	<0.000050	0.00049	0.0016	0.0031	0.776	<0.000020	<2.0	
W7	6/30/2005	0.00049	0.0641	<0.00050	<0.00050	<0.010	29.7	<0.000050	0.00024	0.00065	0.00162	0.358		<2.0	
W7	7/29/2005	0.00055	0.07	<0.00050	<0.00050	<0.010	42.4	<0.000050	0.00042	0.00127	0.0101	0.894		<2.0	
W7	8/30/2005	0.00053	0.064	<0.0010		<0.10	27.8	<0.000017	<0.00030	<0.0010	0.0015	0.363	<0.000020	<2.0	
W7	9/28/2005	0.00149	0.113	<0.0010		<0.10	27.8	0.000061	0.00233	0.0037	0.0061	3.42	<0.000020	<2.0	
W7	10/15/2005	0.00052	0.053	<0.00050	<0.00050	<0.010	25.5	<0.000050	0.00018	0.00062	0.00147	0.37		<2.0	
W7	6/2/2006	0.00496	0.264	<0.0005	<0.0005	<0.01	23.9	0.000237	0.00916	0.0234	0.0225	17.2		<2	
W7	6/8/2006	0.00078	0.0586	<0.0005	<0.0005	<0.01	20.4	<0.00005	0.00078	0.0016	0.00403	1.09		<2	

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	TT As (mg/L)	Ba (mg/L)	Total Be (mg/L)	Total Bi (mg/L)	Total B (mg/L)	Total Ca (mg/L)	Total Cd (mg/L)	Total Co (mg/L)	Total Cr (mg/L)	Total Cu (mg/L)	Total Fe (mg/L)	Total Hg (mg/L)	Total K (mg/L)	Total La (mg/L)
W7	6/15/2006														
W7	6/15/2006	0.0008	0.082	<0.0001	<0.0005	0.006	25.4	0.00002	0.0011	0.0032	0.004	1.7		1.1	
W7	6/23/2006	0.0022	0.167	0.0002	<0.0005	0.005	34.5	0.00008	0.0037	0.0091	0.01	7.2		1.6	
W7	6/28/2006	0.001	0.09	<0.0001	<0.0005	0.006	27.4	0.00003	0.001	0.0031	0.004	2.5		1.1	
W7	7/7/2006	0.0004	0.07	<0.0001	<0.0005	0.005	31.2	<0.00001	<0.0001	<0.0005	0.001	0.2		1	
W7	7/12/2006	0.0004	0.067	<0.0001	<0.0005	0.005	31	<0.00001	<0.0001	<0.0005	0.001	0.2		1	
W7	7/20/2006	0.0005	0.065	<0.0001	<0.0005	0.029	31.8	<0.00001	<0.0001	<0.0005	0.001	0.2		0.9	
W7	7/26/2006														
W7	7/26/2006	0.0005	0.072	<0.0001	<0.0005	0.004	33.9	<0.00001	0.0001	<0.0005	0.001	0.3		1.1	
W7	8/2/2006	0.0006	0.07	<0.0001	<0.0005	0.006	32.9	<0.00001	<0.0001	<0.0005	0.001	0.1		1	
W7	8/10/2006	0.0005	0.072	<0.0001	<0.0005	0.005	34.6	<0.00001	0.0004	<0.0005	<0.001	0.1		1.1	
W7	8/25/2006														
W7	8/25/2006	0.0004	0.069	<0.0002	<0.001	0.007	31.8	<0.00002	<0.0002	<0.001	<0.002	<0.2		1.2	
W7	8/30/2006	0.0005	0.07	<0.0001	<0.0005	0.005	30.8	<0.00001	<0.0001	0.0008	<0.001	0.1		1	
W7	9/6/2006	0.0004	0.07	<0.0001	<0.0005	0.006	31.1	<0.00001	0.0001	0.0008	0.001	0.1		1	
W7	9/13/2006	<0.0002	0.037	<0.0001	<0.0005	0.004	32.2	<0.00001	<0.0001	<0.0005	<0.001	<0.2		0.9	
W7	9/20/2006	0.0005	0.07	<0.0001	<0.0005	0.006	32.4	<0.00001	0.0001	<0.0005	0.001	0.2		0.9	
W7	9/28/2006	<0.0004	0.073	<0.0002	<0.001	0.007	32.5	<0.00002	<0.0002	<0.001	<0.002	<0.2		1.1	
W7	10/4/2006	0.0004	0.074	<0.0002	<0.001	0.005	31.6	<0.00002	<0.0002	<0.001	<0.002	<0.2		0.9	
W7	10/12/2006	0.0004	0.076	<0.0001	<0.0005	0.006	33.5	<0.00001	0.0002	0.0007	0.002	0.5		1	
W7	6/5/2007	0.0008	0.084	<0.0001	<0.0005	0.009	26.4	0.00003	0.0012	0.0031	0.003	2	<0.0001	1.1	
W7	6/5/2007	0.0007	0.095	<0.0002	<0.001	0.01	28.1	0.00003	0.001	0.0037	0.003	2.3	<0.0001	1	
W7	6/5/2007	0.001	0.09	<0.0001	<0.0005	0.008	27.2	0.00003	0.0013	0.0036	0.004	2.5	<0.0001	1.3	
W7	6/20/2007	0.0004	0.066	<0.0001	<0.0005	0.007	30.2	<0.00001	0.0002	0.0005	0.002	0.2	<0.0001	1	
W7	7/18/2007	0.0005	0.071	<0.0001	<0.0005	0.006	32.2	0.00002	0.0002	0.0008	0.002	0.3	<0.0001	1	
W7	7/18/2007	0.0003	0.073	<0.0001	<0.0005	0.006	31.9	0.00001	0.0002	0.0007	0.001	0.3	<0.0001	0.9	
W7	7/18/2007	<0.0002	0.072	<0.0001	<0.0005	0.006	31.8	<0.00001	0.0001	0.0007	0.001	0.3	<0.0001	0.9	
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007	0.0003	0.075	<0.0001	<0.0005	0.006	31.2	<0.00001	0.0002	0.0008	0.002	0.2	<0.0001	1	
W7	8/24/2007	0.0004	0.074	<0.0001	<0.0005	0.006	31.5	<0.00001	0.0001	0.0005	0.001	0.2	<0.0001	1	
W7	8/24/2007	0.0004	0.077	<0.0001	<0.0005	0.006	32.9	<0.00001	0.0001	0.0005	0.001	0.2	<0.0001	1.1	
W7	9/18/2007	0.0005	0.067	<0.0001	<0.0005	0.005	27	<0.00001	0.0003	0.0011	0.003	0.5	<0.0001	0.9	
W7	10/30/2007	0.0004	0.076	<0.0001	<0.0005	0.007	33.5	0.00004	0.0003	0.0007	0.012	0.3	<0.0001	1.6	
W7	10/30/2007	0.0002	0.075	<0.0001	<0.0005	0.006	33	0.00002	0.0001	0.0008	0.005	0.2	<0.0001	1.1	
W7	10/30/2007	<0.0002	0.074	<0.0001	<0.0005	0.006	33.6	0.00002	0.0001	0.0007	0.003	0.2	<0.0001	1.1	
W7	10/30/2007														
W7	4/22/2008	0.0005	0.073	<0.00004		0.006	40.2	<0.00007	0.00007	<0.0005	0.004	0.05	<0.00001	2.56	
W7	4/22/2008	0.0012	0.077	<0.00004		0.006	43.2	<0.00007	0.00009	<0.0005	0.004	0.05	<0.00001	2.77	
W7	4/22/2008	0.0006	0.078	<0.00004		0.006	42.4	<0.00007	0.0001	<0.0005	0.004	0.06	<0.00001	2.7	
W7	4/22/2008														
W7	6/3/2008	0.0007	0.061	<0.00004		<0.005	24.4	<0.00007	0.0004	0.0028	0.002	0.78	<0.00001	0.94	
W7	6/3/2008	0.0013	0.06	<0.00004		<0.005	24.2	<0.00007	0.00041	0.0028	0.001	0.71	<0.00001	0.93	
W7	6/3/2008	0.0008	0.062	0.00004		<0.005	25.2	<0.00007	0.00042	0.003	0.002	0.79	<0.00001	0.94	
W7	6/3/2008														
W7	6/17/2008	0.0006	0.064	<0.00004		<0.005	29.2	<0.00008	0.00028	0.0012	0.001	0.4		1.01	
W7	8/6/2008	0.0006	0.083	<0.0001	<0.0005	0.008	31.5	0.00002	0.0006	0.002	0.002	1.02	<0.0001	1	
W7	8/6/2008	0.0007	0.088	<0.0001	<0.0005	0.007	31.4	0.00002	0.0007	0.0026	0.003	1.36	<0.0001	1.1	
W7	8/6/2008	0.0008	0.093	<0.0001	<0.0005	0.008	31.8	0.00002	0.0008	0.0025	0.003	1.46	<0.0001	1.1	
W7	8/6/2008														
W7	9/18/2008	0.0005	0.06	<0.00004	<0.0001	<0.005	23	0.00002	0.00046	0.0022	0.003	0.91	<0.00001	0.68	
W7	10/28/2008	0.0004	0.057	<0.00004	<0.0001	<0.005	32.4	<0.00001	0.00015	0.0013	0.004	0.34	<0.00001	0.9	
W7	10/28/2008	0.0005	0.06	<0.00004	<0.0001	<0.005	32.3	0.00002	0.00011	0.0008	0.004	0.24	<0.00001	0.9	
W7	10/28/2008	0.0004	0.057	<0.00004	<0.0001	<0.005	32.4	<0.00001	0.0001	0.0005	0.003	0.17	<0.00001	0.8	
W7	10/28/2008														
W7	4/30/2009	<0.0002	0.012	<0.00004	<0.0001	<0.005	5.54	0.00008	0.0001	0.0012	0.013	0.19	0.00002	1.5	
W7	6/8/2009	0.0005	0.069	0.00004	<0.0001	0.005	27	0.00001	0.00052	0.0014	0.003	0.92	<0.00001	0.9	
W7	7/28/2009														
W7	8/6/2009	0.0007	0.081	<0.00004	<0.001	<0.005	35.1	<0.00001	0.00013	0.0009	0.001	0.126		1.1	
W7	8/24/2009														
W7	8/25/2009														
W7	8/31/2009														
W7	10/7/2009														
W7	10/7/2009	0.0004	0.067	<0.00004	<0.001	0.008	34.2	0.00091	0.00022	0.0014	0.006	0.357	<0.00001	0.7	
W7	10/25/2009	0.0004	0.077	<0.0001	<0.001	<0.05	35.7	0.00001	<0.0005	<0.001	0.0021	0.141	<0.00002	1.02	
W7	10/25/2009														
W7	4/19/2010	0.0004	0.04	<0.00004	<0.001	0.006	20.5	0.00001	0.0001	0.0004	0.006	0.104	<0.00001	2.1	
W7	4/27/2010	0.0003	0.023	<0.0001	<0.001	<0.05	9.35	0.00003	<0.0005	<0.001	0.0055	0.211	<0.00002	1.01	
W7	4/29/2010	0.0004	0.025	<0.0001	<0.001	<0.05	9.54	0.00002	<0.0005	<0.001	0.0037	0.259	<0.00002	0.9	
W7	5/1/2010	0.0006	0.035	<0.0001	<0.001	<0.05	11.5	0.00002	<0.0005	0.002	0.0074	0.902	<0.00002	0.81	
W7	5/3/2010	0.0007	0.035	<0.0001	<0.001	<0.05	12.3	0.00003	<0.0005	0.001	0.0045	0.983	<0.00002	0.72	
W7	5/5/2010	0.0007	0.039	<0.0001	<0.001	<0.05	14.3	0.00004	<0.0005	0.001	0.0058	1.14	<0.00002	0.72	
W7	5/7/2010	0.0005	0.041	<0.0001	<0.001	<0.05	14.7	0.00003	<0.0005	<0.001	0.0045	0.92	<0.00002	0.68	
W7	5/9/2010	0.0004	0.043	<0.0001	<0.001	<0.05	18.2	0.00005	<0.0005	0.001	0.0147	0.468	<0.00002	0.74	
W7	5/10/2010	0.0005	0.055	<0.0001	<0.001	<0.05	23.3	0.00011	<0.0005	<0.001	0.0059	0.452	<0.00002	0.91	
W7	5/12/2010										<.03	0.19			
W7	5/12/2010	0.0004	0.048	<0.0001	<0.001	<0.05	22.1	0.00009	<0.0005	<0.001	0.0075	0.378	<0.00002	0.98	
W7	5/14/2010	0.0004	0.046	<0.0001	<0.001	<0.05	18.2	0.00005	<0.0005	<0.001	0.0055	0.556	<0.00002	0.71	
W7	5/16/2010	0.0004	0.044	<0.0001	<0.001	<0.05	19.8	0.00005	<0.0005	0.001	0.0031	0.369	<0.0002	0.72	
W7	5/17/2010	0.0005	0.046	<0.0001	<0.001	<0.05	19.6	0.00001	<0.0005	<0.001	0.0035	0.405	<0.0002	0.77	
W7	5/19/2010	<0.001	0.057	<0.0005	<0.001	<0.05	25.6	0.00001	<0.0005	0.001	0.0049	0.248	<0.00002	0.96	
W7	5/21/2010	0.0005	0.06	<0.0002	<0.001	<0.1	24	<0.0001	<0.0005	<0.002	0.003	0.628	<0.0002	<1	
W7	5/23/2010	0.0014	0.102	<0.0002	<0.001	<0.1	25	<0.0001	0.0019	0.004	0.008	3.49	<0.0002	1	
W7	5/25/2010	0.002	0.121	<0.0005	<0.001	<0.1	28	<0.0001	0.0028	0.01	0.013	5.54	<0.00002	2	
W7	5/27/2010	0.006	0.196	<0.0005	<0.001	<0.1	31	0.0002	0.0077	0.025	0.027	15.6	<0.00002	3	
W7	5/29/2010	0.0025	0.131	0.0002	<0.001	<0.1	31	<0.0001	0.0038	0.01	0.013	6.13	<0.00002	2	
W7	5/31/2010	0.0027	0.149	0.0002	<0.001	<0.1	31	0.0001	0.004	0.011	0.013	6.96	<0.00002	2	
W7	6/2/2010	0.0016	0.098	<0.0002	<0.001	<0.1	33	<0.0001	0.0024	0.008	0.008	4.39	<0.0002	2	
W7	6/4/2010	0.0012	0.09	<0.0002	<0.001	<0.1	31	<0.0001	0.0013	0.004	0.009	2.42	<0.0002	1	
W7	6/6/2010	0.0013	0.109	<0.0002	<0.001	<0.1	33	<0.0001	0.0015	0.006	0.006	3.12		2	
W7	6/14/2010	0.0009	0.099	<0.0002	<0.001	<0.1	31	<0.0001	0.001	0.002	0.004	1.62	<0.0002	1	
W7	6/16/2010	0.0005	0.078	<0.0002	<0.001	<0.1	31	<0.0001	<0.0005	<0.002	0.002	0.573	<0.0002	1	
W7	6/18/2010	0.0006	0.078	<0.0002	<0.001	<0.1	30	<0.0001	<0.0005	<0.002	0.002	0.768	<0.0002	1	
W7	6/20/2010	0.0009	0.085	<0.0002	<0.001	<0.1	29	<0.0001	0.0009	0.003	0.006	1.74	<0.0002	1	
W7	6/22/2010	0.0029	0.164	0.0003	<0.001	<0.1	28	0.0001	0.0048	0.012	0.014	8.61	<0.0002	2	
W7	6/24/2010	0.0013	0.078	<0.0002	<0.001	<0.1	27	<0.0001	0.0014	0.003	0.005	2.05	<0.0002	1	
W7	6/26/2010	0.0005	0.066	<0.0002	<0.001	<0.1	26	<0.0001	<0.0005	<0.002	0.002	0.761	<0.0002	<1	
W7	6/28/2010	0.0005	0.064	<0.0002	<0.001	<0.1	28	<0.0001	<0.0005	<0.002	0.002	0.491	<0.0002	<1	
W7	7/5/2010	0.0011	0.086	<0.0002	<0.001	<0.1	27	<0.0001	0.0012	0.003	0.01	2.4	<0.0002	<1	
W7	8/5/2010	0.0005	0.07	<0.0002	<0.001	<0.1	29	<0.0001	<0.0005	<0.002	0.003	0.428	<0.0002	<1	
W7	9/12/2010	0.0005	0.061	<0.0002	<0.001	<0.1	27	<0.0001	<0.0005	<0.002	0.002	0.682	<0.0002	<1	
W7	9/12/2010	0.0005	0.061	<0.0002	<0.001	<0.1	27	<0.0001	<0.0005	<0.002	0.003	0.594	<0.0002	<1	
W7	10/8/2010	0.0002	0.068	<0.0001	<0.001	<0.05	32.2	0.00003	<0.0005	<0.001	0.0017	0.26	<0.00002	0.77	
W7	4/27/2011	0.0003	0.024	<0.0001	<0.001	<0.05	10.9	0.00005	<0.0005	<0.001	0.0119	0.187	<0.00002	1.88	
W7	5/2/2011	0.0003	0.024	<0.0001	<0.001	<0.05	8.66	0.00005	<0.0005	<0.001	0.0285	0.452	<0.00002	1.46	
W7	6/1/2011	0.0005	0.061	<0.0001	<0.001	<0.05	22.9	0.00002	<0.0005	0.001	0.0027	0.775	<0.00002	0.86	
W7	6/16/2011	0.0014	0.096	<0.0001	<0.001	<0.05	26.4	0.00008	0.0021	0.005	0.0075	4.02	<0.00002	1.04	
W7	7/1/2011	0.0025	0.135	0.0002	<0.001	<0.05	22.2	0.00022	0.0035	0.009	0.0127	6.28	<0.00002	0.76	

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	TT As (mg/L)	Ba (mg/L)	Total Be (mg/L)	Total Bi (mg/L)	Total B (mg/L)	Total Ca (mg/L)	Total Cd (mg/L)	Total Co (mg/L)	Total Cr (mg/L)	Total Cu (mg/L)	Total Fe (mg/L)	Total Hg (mg/L)	Total K (mg/L)	Total La (mg/L)
W7	8/2/2011	0.0037	0.211	0.0003	<0.001	<0.05	20.7	0.00017	0.0064	0.021	0.0197	13.8	<0.00002	1.11	
W7	8/16/2011	0.001	0.069	<0.0001	<0.001	<0.05	25.4	0.00004	0.0006	0.001	0.0143	1.25	<0.00002	0.53	
W7	8/19/2011	0.0009	0.061	<0.0001	<0.001	<0.05	24.1	0.00002	0.0006	0.001	0.0028	1.21	<0.00002	0.47	
W7	9/12/2011	0.0008	0.061	<0.0001	<0.001	<0.05	30.1	0.00005	<0.0005	<0.001	0.0024	0.738	<0.00002	0.6	
W7	10/6/2011	0.0005	0.058	<0.0001	<0.001	<0.05	28.4	0.00001	<0.0005	<0.001	0.0012	0.49	<0.00002	0.62	
W7	11/22/2011	0.0006	0.085	<0.0001	<0.001	<0.05	36.4	0.00025	<0.0005	0.001	0.0108	0.865	<0.000020	1.09	
W7	1/24/2012	0.0004	0.087	<0.0001	<0.001	<0.05	39.5	0.00001	<0.0005	<0.001	0.0013	0.113	<0.000010	1.57	
W7	2/25/2012	0.0003	0.082	<0.0001	<0.001	<0.05	41	0.00003	<0.0005	<0.001	0.0055	0.043	<0.000010	1.51	
W7	3/5/2012	0.0004	0.083	<0.0001	<0.001	<0.05	39.7	0.00001	<0.0005	<0.001	0.0068	0.108	<0.000010	1.46	
W7	4/24/2012	0.00074	0.0404	<0.00010	<0.0010	<0.05	9.61	0.000029	0.00056	0.0015	0.00635	1.32	<0.000010	1.2	
W7	5/6/2012	0.00045	0.0317	<0.00010	<0.0010	<0.05	11.2	0.000015	<0.00050	<0.0010	0.00299	0.551	<0.000010	0.881	
W7	5/6/2012	0.00049	0.0316	<0.00010	<0.0010	<0.05	11.1	0.000016	<0.00050	<0.0010	0.00373	0.572	<0.000010	0.912	
W7	6/19/2012	0.00232	0.154	0.00016	<0.0010	<0.05	26.4	0.000119	0.00326	0.0112	0.0102	7.88	<0.000010	1.09	
W7	7/17/2012	0.00118	0.0843	<0.00010	<0.0010	<0.05	31.7	0.000034	0.00075	0.0024	0.0035	2.28	<0.000010	0.824	
W7	7/30/2012	0.00095	0.0737	<0.00010	<0.0010	<0.05	32.1	0.000014	<0.00050	<0.0010	0.0045	1.32	<0.000010	1.14	
W7	8/8/2012	0.00085	0.0731	<0.00010	<0.0010	<0.05	35.7	<0.000010	<0.00050	<0.0010	0.00108	1.13	<0.000010	0.894	
W7	8/23/2012	0.00093	0.0778	<0.00010	<0.0010	<0.05	39.1	<0.000010	<0.00050	<0.0010	0.00134	1.12	<0.000010	0.929	
W7	9/13/2012	0.0007	0.0629	<0.00010	<0.0010	<0.05	31.2	<0.000010	<0.00050	<0.0010	0.00208	1.15	<0.000010	0.589	
W7	10/16/2012	0.00065	0.0693	<0.00010	<0.0010	<0.05	36	<0.000010	<0.00050	<0.0010	0.00144	0.995	<0.000010	0.748	
W7	11/2/2012	0.00058	0.0848	<0.00010	<0.0010	<0.05	37.3	0.000015	<0.00050	<0.0010	0.00231	0.289	<0.000010	1.16	
W7	12/29/2012	0.00042	0.104	<0.00010	<0.0010	<0.05	49.6	0.000015	<0.00050	<0.0010	0.0028	0.0724	<0.000010	2.29	
W7	4/30/2013	0.00019	0.0996	<0.00010	<0.0010	<0.05	47.8	<0.000010	<0.00050	<0.0010	0.0021	0.0288	<0.000010	1.95	
W7	5/14/2013	0.00072	0.0462	<0.00010	<0.0010	<0.05	9.61	0.000025	0.00107	0.0027	0.0084	2.2	<0.000010	1.18	
W7	6/19/2013	0.00064	0.0691	<0.00010	<0.0010	<0.05	29.1	0.000011	<0.00050	<0.0010	0.00222	0.702	<0.000010	0.847	
W7	6/19/2013	0.00058	0.0683	<0.00010	<0.0010	<0.05	29.7	<0.000010	<0.00050	<0.0010	0.00191	0.657	<0.000010	0.872	
W7	7/13/2013	0.00063	0.0712	<0.00010	<0.0010	<0.05	35.1	<0.000010	<0.00050	<0.0010	0.00143	0.631	<0.000010	0.856	
W7	8/21/2013	0.0007	0.0755	<0.00010	<0.0010	<0.05	35.8	<0.000010	<0.00050	<0.0010	0.00138	0.619	<0.000010	0.951	
W7	9/6/2013	0.00099	0.0815	<0.00010	<0.0010	<0.05	38.7	<0.000010	<0.00050	0.0014	0.0017	1.48	<0.000010	0.968	
W7	10/21/2013	0.00061	0.0642	<0.00010	<0.0010	<0.05	31.8	<0.000010	<0.00050	<0.0010	0.00119	0.683	<0.000010	0.754	
W7	11/16/2013	0.00052	0.0832	<0.00010	<0.0010	<0.05	39.8	<0.000010	<0.00050	<0.0010	0.00124	0.296	<0.000010	1.14	
W7	12/7/2013	0.00059	0.0967	<0.00010	<0.0010	<0.05	46.7	0.000013	<0.00050	<0.0010	0.00141	0.227	<0.000010	1.52	
C4	5/6/2012	0.0125	0.56	0.00088	<0.0010	<0.05	23.9	0.000416	0.0178	0.06	0.0585	45.8	<0.000010	3.02	
C4	8/1/2012	0.00497	0.24	0.00031	<0.0010	<0.05	34.3	0.000114	0.00591	0.0135	0.0167	13.9	<0.000010	1.34	
C4	8/8/2012	0.00635	0.261	0.00033	<0.0010	<0.05	36.5	0.000107	0.00701	0.0148	0.015	16.1	<0.000010	1.45	
C4	8/25/2012	0.00759	0.373	0.00042	<0.0010	<0.05	45.6	0.000202	0.00894	0.0079	0.0218	18.2	<0.000010	1.19	
C4	6/22/2013	0.0123	0.539	0.00075	<0.0010	<0.05	55.7	0.000317	0.0172	0.0426	0.0479	36.2	0.000018	2.77	
C4	7/19/2013	0.00864	0.373	0.00048	<0.0010	<0.05	54	0.000209	0.0113	0.0287	0.0323	24.4	0.000015	2.21	
C4	8/22/2013	0.00939	0.438	0.00055	<0.0010	<0.05	51	0.000308	0.0133	0.0268	0.0347	26.2	0.000033	2.02	
C4	9/6/2013	0.00462	0.189	0.00019	<0.0010	<0.05	43.1	0.000066	0.00447	0.0107	0.0103	11	<0.000010	1.5	
C4	10/18/2013	0.00153	0.0845	<0.00010	<0.0010	<0.05	37.3	0.000017	0.00102	0.0012	0.0019	2.12	<0.000010	0.858	
C4	10/18/2013	0.00151	0.0837	<0.00010	<0.0010	<0.05	36.9	0.000012	0.00101	0.0013	0.00263	2.1	<0.000010	0.833	
C4	11/17/2013	0.00545	0.17	<0.00010	<0.0010	<0.05	70.5	0.000022	0.00621	0.0015	0.00357	14.7	<0.000010	1.05	
C10	5/12/2012	0.0208	0.942	0.00176	<0.0010	<0.05	51.2	0.000982	0.0315	0.0779	0.1	73.2	<0.000010	4.64	
C10	7/30/2012	0.0123	0.549	0.0008	<0.0010	<0.05	56.3	0.000365	0.0151	0.0385	0.0527	37.5	0.000062	3.36	
C10	7/30/2012														
C10	8/10/2012	0.0124	0.666	0.00102	<0.0010	<0.05	67.1	0.000486	0.0185	0.0378	0.0597	38.9	<0.000010	3.11	
C10	8/27/2012	0.0156	0.754	0.00117	<0.0010	<0.05	72.2	0.000526	0.0213	0.0455	0.0624	45.6	<0.000010	3.37	
C10	6/22/2013	0.00203	0.0981	<0.00010	<0.0010	<0.05	44.4	0.000019	0.00109	0.0022	0.00492	3.18	<0.000010	1.46	
C10	7/23/2013	0.00386	0.174	0.00022	<0.0010	<0.05	45.1	0.000071	0.00371	0.0095	0.0114	9.35	<0.000010	1.71	
C10	8/22/2013	0.00213	0.1	<0.00010	<0.0010	<0.05	49.5	0.000016	0.00099	0.0019	0.00344	2.95	<0.000010	1.18	
C10	9/8/2013	0.00188	0.0869	<0.00010	<0.0010	<0.05	48.6	0.000019	0.00094	0.0013	0.00241	2.71	<0.000010	1.17	
C10	9/8/2013	0.00194	0.0852	<0.00010	<0.0010	<0.05	48	0.000093	0.00086	0.0012	0.00237	2.64	<0.000010	1.17	
C10	10/18/2013	0.00177	0.091	<0.00010	<0.0010	<0.05	46.8	0.00002	0.00139	0.0024	0.00355	3.33	<0.000010	1.1	

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Total Li (mg/L)	Total Mg (mg/L)	Total Mn (mg/L)	Total Mo (mg/L)	Total Na (mg/L)	Total Ni (mg/L)	Total Pb (mg/L)	Total P (mg/L)	Total Sb (mg/L)	Total Se (mg/L)	Total Si (mg/L)	Total Sn (mg/L)	Total Sr (mg/L)	Total S (mg/L)
W1	5/27/2005	<0.0050	8.65	0.0612	0.0011	4.9	0.0032	<0.00050		<0.00050	<0.0010		<0.00050		
W1	6/30/2005	<0.0050	12.3	0.00663	0.00107	8.6	0.00101	<0.000050	<0.30	<0.00010	<0.0010	5.43	<0.00010	0.293	
W1	7/28/2005	<0.0050	12.7	0.0164	0.00153	8.5	0.00113	<0.000050	<0.30	<0.00010	<0.0010	5.21	<0.00010	0.333	
W1	8/29/2005	<0.0050	11.5	0.0258	0.001	6.8	0.0016	<0.00050		<0.00050	<0.0010		<0.00050		
W1	9/28/2005	<0.0050	11.3	0.0559	<0.0010	6.3	0.0019	<0.00050		<0.00050	<0.0010		<0.00050		
W1	10/15/2005	<0.0050	10.9	0.0483	0.000867	5.9	0.0024	0.00012	<0.30	<0.00010	<0.0010	5.99	<0.00010	0.269	
W2	5/27/2005	<0.0050	8.37	0.0505	<0.0010	4.8	0.0026	<0.00050		<0.00050	<0.0010		<0.00050		
W2	5/27/2005	<0.005	5.78	0.0536			0.0033	0.00066		<0.0005					
W2	6/30/2005	<0.0050	12.5	0.00785	0.00103	8.7	0.00105	<0.000050	<0.30	<0.00010	<0.0010	5.48	<0.00010	0.304	
W2	7/28/2005	<0.0050	13.1	0.014	0.00134	8.7	0.00107	<0.000050	<0.30	<0.00010	<0.0010	5.32	<0.00010	0.338	
W2	8/30/2005	<0.0050	11.6	0.025	0.0011	6.9	0.0016	<0.00050		<0.00050	<0.0010		<0.00050		
W2	9/28/2005	<0.0050	11	0.0543	<0.0010	6	0.002	<0.00050		<0.00050	<0.0010		<0.00050		
W2	10/15/2005	<0.0050	10.8	0.0368	0.000916	6	0.00164	0.000078	<0.30	<0.00010	<0.0010	5.95	<0.00010	0.267	
W3	5/27/2005	<0.0050	9.03	0.023	0.0011	4.9	0.0017	<0.00050		<0.00050	<0.0010		<0.00050		
W3	6/30/2005	<0.0050	17.6	0.0084	0.00163	12.1	0.00087	0.000292	<0.30	<0.00010	<0.0010	6.04	<0.00010	0.299	
W3	7/29/2005	<0.0050	17.2	0.0641	0.0019	9.4	0.00257	0.000359	<0.30	0.0001	<0.0010	7.03	0.00011	0.295	
W3	8/29/2005	<0.0050	13.2	0.0806	0.0014	7.8	0.0017	<0.00050		<0.00050	<0.0010		<0.00050		
W3	9/28/2005	<0.0050	12	0.12	0.0012	7.1	0.0018	<0.00050		<0.00050	<0.0010		<0.00050		
W3	10/15/2005	<0.0050	12.5	0.124	0.00126	7	0.00136	<0.000050	<0.30	<0.00010	<0.0010	5.86	<0.00010	0.233	
W8	5/27/2005	<0.0050	5.03	0.00512	<0.0010	2.6	0.0018	<0.00050		<0.00050	<0.0010		<0.00050		
W8	6/30/2005	<0.0050	7.57	0.0445	0.000512	4.5	0.00172	0.000066	<0.30	<0.00010	<0.0010	5.21	<0.00010	0.112	
W8	7/29/2005	<0.0050	9.43	0.0268	0.000565	3.9	0.00146	0.000059	<0.30	<0.00010	<0.0010	5.04	<0.00010	0.133	
W8	8/30/2005	<0.0050	8.04	0.0447	<0.0010	3.8	0.0018	<0.00050		<0.00050	<0.0010		<0.00050		
W8	9/28/2005	<0.0050	7.45	0.0487	<0.0010	3.7	0.0017	<0.00050		<0.00050	<0.0010		<0.00050		
W8	10/15/2005	<0.0050	7.41	0.24	0.00171	4.5	0.00109	0.000064	<0.30	<0.00010	<0.0010	6.29	<0.00010	0.176	
W9	5/27/2005	<0.0050	4.1	0.00703	<0.0010	2.3	0.0013	<0.00050		<0.00050	<0.0010		<0.00050		
W9	6/30/2005	<0.0050	6.04	0.0432	0.000356	4.2	0.00125	<0.000050	<0.30	<0.00010	<0.0010	5.32	<0.00010	0.0956	
W9	7/29/2005	<0.0050	7.18	0.0638	0.00112	3.8	0.00152	0.000099	<0.30	<0.00010	<0.0010	5.5	<0.00010	0.146	
W9	8/29/2005	<0.0050	6	0.0669	<0.0010	3.5	0.0013	<0.00050		<0.00050	<0.0010		<0.00050		
W9	9/28/2005	<0.0050	5.52	0.0485	<0.0010	3.1	0.0012	<0.00050		<0.00050	<0.0010		<0.00050		
W9	10/15/2005	<0.0050	5.17	0.0363	0.000349	2.9	0.00105	<0.000050	<0.30	<0.00010	<0.0010	5.67	<0.00010	0.0837	
W10	5/27/2005	<0.0050	3.25	0.00326	<0.0010	<2.0	0.0018	<0.00050		<0.00050	<0.0010		<0.00050		
W10	5/27/2005	<0.005			<0.001	<2	0.0033	<0.0005			<0.001		<0.0005		
W10	6/30/2005	<0.0050	3.71	0.0457	0.000177	3.6	0.00191	0.000388	<0.30	<0.00010	<0.0010	5.43	<0.00010	0.0778	
W10	7/29/2005	<0.0050	5	0.174	0.000273	4.7	0.00411	0.00178	<0.30	0.00015	<0.0010	9.73	0.00026	0.108	
W10	8/29/2005	<0.0050	6.02	0.0422	<0.0010	5.3	0.0016	<0.00050		<0.00050	<0.0010		<0.00050		
W10	9/28/2005	<0.0050	8.84	0.00236	<0.0010	5.4	<0.0010	<0.00050		<0.00050	<0.0010		<0.00050		
W10	10/15/2005	<0.0050	8.41	0.00586	0.000277	4.4	0.00086	<0.000050	<0.30	<0.00010	<0.0010	5.52	<0.00010	0.153	
W6	5/27/2005	<0.0050	4.88	0.0169	<0.0010	2.7	0.0021	<0.00050		<0.00050	<0.0010		<0.00050		
W6	6/30/2005	<0.0050	6.65	0.0113	0.000448	4.6	0.00135	0.000134	<0.30	<0.00010	<0.0010	7.27	<0.00010	0.1	
W6	7/29/2005	<0.0050	15.4	0.0536	0.00133	8.4	0.00204	0.00035	<0.30	<0.00010	<0.0010	6.69	0.00013	0.286	
W6	8/30/2005	<0.0050	7.97	0.0224	<0.0010	4.5	0.0011	<0.00050		<0.00050	<0.0010		<0.00050		
W6	9/28/2005	<0.0050	8.02	0.0267	<0.0010	4.2	0.0013	<0.00050		<0.00050	<0.0010		<0.00050		
W6	10/15/2005	<0.0050	7.18	0.0188	0.000392	4	0.00131	<0.000050	<0.30	<0.00010	<0.0010	6.54	<0.00010	0.112	
W6	6/2/2006	<0.005	5.63	0.0835	0.000411	3.4	0.00535	0.000851	<0.3	0.00017	<0.001	7.95	0.0001	0.0816	
W6	8/25/2006	<0.002	7.8	0.04	<0.002	4.8	0.001	<0.0002		<0.0004	<0.0004	7.39	<0.002	0.14	0.9
W6	9/6/2006	0.001	7.3	0.042	<0.001	4.6	0.0015	0.0001		<0.0002	0.0003	7.42	<0.001	0.135	1
W6	6/20/2007	<0.001	5.6	0.011	<0.001	4.3	0.0017	<0.0001	0.18	<0.0002	0.0002	6.4	<0.001	0.104	0.9
W6	9/18/2007	0.001	6.9	0.023	<0.001	5	0.0012	<0.0001	<0.05	<0.0002	<0.0002	7.2	<0.001	0.125	1
W6	6/17/2008	<0.001	6.44	0.0215	0.00045	4.8	0.001	<0.0001	0.11	<0.0002	<0.0006	5.34	<0.0001	0.103	
W6	9/18/2008	<0.001	6.61	0.0098	0.00031	4.1	0.001	<0.0001	0.02	<0.0002	<0.0006	6.22	<0.0001	0.104	
W6	4/14/2009	0.005	50	0.083	0.00334	33.3	0.001	0.0011	0.34	0.0003	0.0017	6.62	<0.0001	0.524	42.4
W6	4/30/2009	0.004	7.92	0.167	0.00043	3.38	0.009	0.0024	0.25	<0.0002	<0.0006	11.5	<0.0001	0.142	2.7
W6	7/28/2009														
W6	8/6/2009	0.002	16.7	0.1	0.0092	15.5	0.002	0.0003	0.017	0.0002	<0.0006	4.67	<0.0001	0.666	14.7
W6	8/6/2009	0.002	16.7	0.0874	0.0088	15.8	0.001	0.0006	0.015	0.0003	0.0007	4.68	<0.0001	0.566	15
W6	8/18/2009														
W6	8/25/2009														
W6	10/7/2009														
W6	10/7/2009	0.002	15.4	0.28	0.016	19.4	0.002	0.0007	<0.05	0.0006	0.0018	3.6	<0.0001	0.842	22.4
W6	10/25/2009	<0.005	16.9	0.229	0.015	18.4	0.001	<0.0002		0.0006	0.0019	4.48	<0.005	0.844	27
W6	4/19/2010	0.002	14.7	0.119	0.0017	9.35	0.002	0.0006	0.08	<0.0002	<0.0006	6.58	<0.0001	0.392	10.9
W6	5/16/2010	<0.005	8.41	0.027	<0.001	4.4	<0.001	<0.0002	0.021	<0.0005	0.0002	5.35	<0.005	0.197	<3
W6	5/17/2010	<0.005	4.84	0.05	<0.001	2.96	<0.001	<0.0002	0.027	<0.0005	<0.0001	2.02	<0.005	0.19	<3
W6	5/19/2010	<0.005	8.01	0.026	<0.001	5.29	<0.003	<0.0002	0.037	<0.0005	0.0002	2.87	<0.005	0.304	4
W6	5/21/2010	<0.01	6	0.092	<0.001	4	<0.001	<0.0002		<0.0005	<0.0008	2.38	<0.005	0.279	<60
W6	5/23/2010	<0.01	7	0.024	<0.001	5	<0.001	<0.0002		<0.0005	<0.0008	2.36	<0.005	0.284	<60
W6	5/25/2010	<0.01	10	0.03	<0.001	8	<0.003	0.0004		<0.0005	<0.001	2.95	<0.005	0.328	<60
W6	5/29/2010	<0.01	4	0.039	<0.001	2	<0.001	<0.0002		<0.0005	<0.0008	<2	<0.005	0.225	<60
W6	6/2/2010	<0.01	22	0.044	0.003	15	<0.001	<0.0002		<0.0005	<0.0008	5.61	<0.005	0.704	<60
W6	6/4/2010	<0.01	7	0.008	<0.001	4	0.001	<0.0002		<0.0005	<0.0008	7.13	<0.005	0.116	<60
W6	6/6/2010	<0.01	8	0.012	<0.001	4	0.001	0.0005		<0.0005	<0.0008	6.82	<0.005	0.121	<60
W6	6/14/2010	<0.01	7	0.014	<0.001	4	0.002	0.0002		<0.0005	<0.0008	7.45	<0.005	0.123	<60
W6	6/16/2010	<0.01	7	0.013	<0.001	4	0.001	<0.0002		<0.0005	<0.0008	7.12	<0.005	0.114	<60
W6	6/18/2010	<0.01	7	0.01	<0.001	4	0.001	<0.0002		<0.0005	<0.0008	6.99	<0.005	0.111	<60
W6	6/20/2010	<0.01	7	0.009	<0.001	4	0.001	<0.0002		<0.0005	<0.0008	7.28	<0.005	0.116	<60
W6	6/22/2010	<0.01	7	0.011	<0.001	4	0.002	<0.0002		<0.0005	<0.0008	8.49	<0.005	0.119	<60
W6	6/24/2010	<0.01	6	0.006	<0.001	3	<0.001	0.0003		<0.0005	<0.0008	5.99	<0.005	0.095	<60
W6	6/26/2010	<0.01	7	0.009	<0.001	4	0.001	<0.0002		<0.0005	<0.0008	6.95	<0.005	0.114	<60
W6	6/26/2010														
W6	6/28/2010	<0.01	6	0.023	<0.001	4	0.002	0.0003		<0.0005	<0.0008	9.39	<0.005	0.117	<60
W6	7/5/2010	<0.01	7	0.053	<0.001	3	0.004	0.0006		<0.0005	<0.0008	8.82	<0.005	0.112	<60
W6	8/5/2010	<0.01	8	0.023	<0.001	4	0.002	<0.0002		<0.0005	<0.0008	7.78	<0.005	0.121	<60
W6	9/9/2010	<0.01	8	0.045	<0.001	4	0.003	0.0004		<0.0005	<0.0008	8.17	<0.005	0.142	<60
W6	10/8/2010	<0.005	8.87	0.043	<0.001	4.58	0.002	0.0002		<0.0005	<0.0001	6.6	<0.005	0.141	<3
W6	8/19/2011	<0.005	20.9	0.02	0.003	14.6	0.001	<0.0002	<0.01	<0.0005	0.0004	4.9	<0.005	0.614	18
W6	7/17/2012	0.0136	14.6	0.771	0.0012	4.89	0.0348	0.00971	0.99	0.00072	0.00068	42.3	<0.0050	0.23	<3.0
W6	7/30/2012	0.0088	12.1	0.513	<0.0010	4.72	0.0236	0.00627	0.642	0.00051	0.00061	33.1	<0.0050	0.202	<3.0
W6	8/9/2012	0.0061	10.7	0.376	<0.0010	4.85	0.0126	0.00343	0.373	<0.00050	0.00041	21.7	<0.0050	0.174	<3.0
W6	8/25/2012	<0.0050	11.7	0.619	<0.0010	4.59	0.0192	0.00505	0.647	<0.00050	0.00016	13.4	<0.0050	0.227	<3.0
W6	9/13/2012	<0.0050	6.92	0.0619	<0.0010	4.05	0.005	0.00025	0.04	<0.00050	<0.00010	6.82	<0.0050	0.12	<3.0
W6	10/13/2012	<0.0050	9.61	0.0661	<0.0010	5.31	0.0016	0.00023	0.048	<0.00050	0.00014	8.91	<0.0050	0.154	3.2
W6	11/12/2012	<0.0050	11.3	0.0021	<0.0010	6.12	0.0013	<0.00020	0.015	<0.00050	0.00011	9.52	<0.0050	0.185	<3.0
W6	5/14/2013	<0.0050	2.5	0.0637	<0.0010	1.07	0.0023	0.00024	0.086	<0.00050	<0.00010	3.97	<0.0050	0.0391	<3.0
W6	6/22/2013	<0.0050	9.52	0.161	<0.0010	4.63	0.0102	0.00184	0.184	<0.00050	0.00022	15.9	<0.0050	0.15	<3.0
W6	7/12/2013	<0.0050	6.98	0.0413	<0.0010	4.18	0.0021	0.00021	0.032	<0.00050	0.00012	7.44	<0.0050	0.129	<3.0
W6	8/22/2013	<0.0050	7.74	0.0358	<0.0010	4.67	0.0013	<0.00020	0.018	<0.00050	<0.00010	8	<0.0050	0.134	<3.0
W6	9/6/2013	<0.0050	8.38	0.0387	<0.0010	4.75	0.0014	<0.00020	0.016	<0.00050	0.00013	8.12	<0.0050	0.142	<3.0
W6	10/7/2013	<0.0050	8	0.0965	<0.0010	4.45	0.0039	0.00081	0.098	<0.00050	0.00015	11.3	<0.0050	0.136	<3.0
W6	11/16/2013	<0.0050	9.65	0.0488	<0.0010	5.26	0.0014	<0.00020	0.022	<0.00050	<0.00010	8.78	<0.0050	0.163	<3.0
W6	12/7/2013	<0.0050	14.5	0.0036	<0.0010	7.78	0.0013	<0.00020	0.031	<0.00050	0.00018	9.08	<0.0050	0.225	<3.0
W7	5/27/2005	<0.0050	6.82	0.0519	0.001	3.9	0.003	<0.00050		<0.00050	<0.0010		<0.00050		
W7	6/30/2005	<0.0050	10.1	0.0514	0.00119	6.8	0.00141	0.000134	<0.30	<0.00010	<0.0010	5.85	<0.00010	0.279	
W7	7/29/2005	<0.0050	14.2	0.0468	0.00121	7.7	0.00174	0.000332	<0.30	<0.00010	<0.0010	6.39	<0.00010	0.287	
W7	8/30/2005	<0.0050	9.67	0.0368	0.0011	5.9	0.0016	<0.00050		<0.00050	<0.0010		<0.00050		
W7	9/28/2005	<0.0050	9.57	0.314	<0.0010	4.9	0.0068	0.0011		<0.00050	<0.0010		<0.00050		
W7	10/15/2005	<0.0050	8.77	0.0473	0.000959	5.1	0.00176	0.000061	<0.30	<0.00010	<0.0010	6.09	<0.00010	0.24	
W7	6/2/2006	0.0079	10.7	0.589	0.000988	4.5	0.0275	0.00547	0.77	0.00037	<0.001	22.4	0.00036	0.204	
W7	6/8/2006	<0.005	7.12	0.0645	0.000843	4.4	0.00402	0.00045	<0.3	<0.0001	<0.001	5.27	<0.0001	0.174	

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Total Li (mg/L)	Total Mg (mg/L)	Total Mn (mg/L)	Total Mo (mg/L)	Total Na (mg/L)	Total Ni (mg/L)	Total Pb (mg/L)	Total P (mg/L)	Total Sb (mg/L)	Total Se (mg/L)	Total Si (mg/L)	Total Sn (mg/L)	Total Sr (mg/L)	Total S (mg/L)
W7	6/15/2006														
W7	6/15/2006	0.002	8.8	0.079	0.001	5.8	0.0052	0.0006		<0.0002	0.0002	7.24	<0.001	0.276	3.2
W7	6/23/2006	0.005	12.5	0.226	0.001	7	0.0117	0.002		<0.0002	0.0004	12.3	<0.001	0.367	4.2
W7	6/28/2006	0.002	9.3	0.086	0.001	6	0.0042	0.0007		<0.0002	<0.0002	8.36	<0.001	0.267	3.1
W7	7/7/2006	0.001	10.1	0.035	0.001	7.4	0.0019	<0.0001		<0.0002	0.0002	5.57	<0.001	0.329	3.7
W7	7/12/2006	0.001	9.5	0.035	0.001	6.5	0.0014	<0.0001		<0.0002	0.0002	5.73	<0.001	0.323	3.8
W7	7/20/2006	0.001	11.2	0.042	0.001	6.4	0.0011	<0.0001		<0.0002	<0.0002	5.72	<0.001	0.334	3.9
W7	7/26/2006														
W7	7/26/2006	0.001	11.7	0.051	0.001	6.5	0.0009	0.0001		<0.0002	<0.0002	6.21	<0.001	0.362	4.1
W7	8/2/2006	0.001	10.8	0.033	0.001	6.9	0.001	<0.0001		<0.0002	0.0005	5.94	<0.001	0.342	3.6
W7	8/10/2006	0.001	11.3	0.04	0.001	7.2	0.0009	<0.0001		<0.0002	<0.0002	5.82	<0.001	0.362	4
W7	8/25/2006														
W7	8/25/2006	<0.002	10	0.039	<0.002	6.7	0.001	<0.0002		<0.0004	<0.0004	5.49	<0.002	0.335	3.3
W7	8/30/2006	0.001	10.3	0.035	0.001	6.7	0.0013	<0.0001		<0.0002	<0.0002	5.49	<0.001	0.309	3.5
W7	9/6/2006	0.001	10.3	0.035	0.001	6.8	0.0013	0.0002		<0.0002	0.0004	5.74	<0.001	0.332	3.7
W7	9/13/2006	<0.001	11	0.033	<0.001	7.1	0.0007	<0.0001		<0.0002	0.0002	6.36	<0.001	0.187	3.6
W7	9/20/2006	0.001	11.7	0.042	0.001	6.6	0.0014	0.0001		<0.0002	0.0004	5.89	<0.001	0.347	3.7
W7	9/28/2006	<0.002	11	0.037	<0.002	6.9	0.002	<0.0002		<0.0004	0.0004	5.35	<0.002	0.328	3.6
W7	10/4/2006	<0.002	11	0.038	<0.002	7	0.002	<0.0002		<0.0004	0.0006	5.75	<0.002	0.347	3.5
W7	10/12/2006	0.001	11.8	0.063	0.001	6.6	0.0015	0.0002		<0.0002	<0.0002	6.51	<0.001	0.35	4
W7	6/5/2007	0.002	10.2	0.126	0.001	5.8	0.0033	0.0006	0.14	<0.0002	<0.0002	7.47	<0.001	0.275	3.4
W7	6/5/2007	<0.002	11	0.14	<0.002	6.1	0.0039	0.0007	0.11	<0.0004	<0.0004	8.37	<0.002	0.302	3.5
W7	6/5/2007	0.002	10.6	0.135	0.001	6	0.0041	0.0007	0.14	<0.0002	<0.0002	8.3	<0.001	0.276	3.5
W7	6/20/2007	0.001	9.8	0.044	0.002	7.2	0.0016	0.0001	<0.05	<0.0002	0.0003	5.45	<0.001	0.304	4.2
W7	7/18/2007	0.001	11	0.03	0.001	7.7	<0.0005	0.0004		<0.0002	0.0004	5.78	<0.001	0.38	4
W7	7/18/2007	0.001	11	0.035	0.001	7.6	0.0009	0.0001		<0.0002	0.0007	5.77	<0.001	0.379	3.9
W7	7/18/2007	0.001	11	0.026	0.001	7.6	0.0005	<0.0001		<0.0002	0.0005	5.73	<0.001	0.37	3.9
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007	0.001	10.7	0.033	0.001	7.9	0.0014	0.0001	<0.05	<0.0002	<0.0002	5.8	<0.001	0.34	3.3
W7	8/24/2007	0.001	10.8	0.03	0.001	7.6	0.0013	<0.0001	<0.05	<0.0002	<0.0002	5.92	<0.001	0.34	3.3
W7	8/24/2007	0.001	11.2	0.031	0.002	8	0.0014	<0.0001	<0.05	<0.0002	<0.0002	6.3	<0.001	0.376	3.5
W7	9/18/2007	0.002	9.5	0.034	0.001	7.3	0.0015	0.0002	<0.05	<0.0002	<0.0002	6.34	<0.001	0.309	2.8
W7	10/30/2007	0.001	11.2	0.054	0.002	8.1	0.0008	0.0023		<0.0002	0.0007	5.94	<0.001	0.395	4.6
W7	10/30/2007	0.001	11	0.046	0.001	7.7	0.001	0.0005		<0.0002	0.0003	5.09	<0.001	0.388	4.3
W7	10/30/2007	0.001	11.3	0.047	0.001	7.6	0.0009	0.0003		<0.0002	<0.0002	5.97	<0.001	0.385	4.2
W7	10/30/2007														
W7	4/22/2008	0.002	19	0.0257	0.00165	11.8	<0.001	<0.0001	0.02	<0.0002	<0.0006	3.8	<0.004	0.487	
W7	4/22/2008	0.002	20.6	0.0268	0.00185	12.4	<0.001	<0.0001	0.02	<0.0002	<0.0006	4.15	<0.004	0.518	
W7	4/22/2008	0.002	20.2	0.0268	0.00174	12	0.001	<0.0001	0.02	<0.0002	<0.0006	4.03	<0.004	0.514	
W7	4/22/2008														
W7	6/3/2008	0.001	8.68	0.0693	0.00131	6.6	0.002	0.0002	0.05	<0.0002	<0.0006	2.66	<0.0001	0.234	
W7	6/3/2008	0.001	8.71	0.0616	0.00129	6.8	0.002	0.0001	0.05	<0.0002	<0.0006	2.68	<0.0001	0.232	
W7	6/3/2008	0.001	9.15	0.065	0.00129	7	0.002	0.0002	0.05	<0.0002	0.0009	2.77	<0.0001	0.24	
W7	6/3/2008														
W7	6/17/2008	<0.001	10.1	0.0833	0.00148	7.5	0.001	0.0001	0.06	<0.0002	0.0008	4.43	<0.0001	0.294	
W7	8/6/2008	0.002	10.8	0.064	0.001	7	0.003	0.0003		<0.0002	<0.0002	6.15	<0.001	0.363	3.5
W7	8/6/2008	0.002	10.8	0.059	0.001	7.4	0.0034	0.0003		<0.0002	<0.0002	6.45	<0.001	0.361	3.5
W7	8/6/2008	0.002	11	0.071	0.001	7.4	0.0036	0.0004		<0.0002	<0.0002	7.39	<0.001	0.36	3.5
W7	8/6/2008														
W7	9/18/2008	<0.001	7.62	0.0347	0.00082	5.1	0.002	0.0003	0.04	<0.0002	0.0008	7.09	<0.0001	0.201	
W7	10/28/2008	<0.001	11.6	0.0347	0.00107	6.52	0.001	0.0001	0.016	<0.0002	<0.0006	7.13	<0.0001	0.263	
W7	10/28/2008	<0.001	11.5	0.0396	0.00114	6.24	0.003	0.0026	0.031	<0.0002	<0.0006	7.08	<0.0001	0.271	
W7	10/28/2008	<0.001	11.5	0.0337	0.00111	6.26	0.004	<0.0001	0.013	<0.0002	<0.0006	7.04	<0.0001	0.274	
W7	10/28/2008														
W7	4/30/2009	<0.001	1.97	0.0153	0.00016	1.29	<0.001	0.0002	0.09	<0.0002	<0.0006	1.67	<0.0001	0.036	0.5
W7	6/8/2009	0.001	9.53	0.0449	0.00141	6.08	0.002	0.0003	<0.05	<0.0002	<0.0006	5.67	<0.0001	0.281	3.2
W7	7/28/2009														
W7	8/6/2009	0.001	12.2	0.0278	0.0015	7.17	0.001	0.0004	0.011	<0.0002	<0.0006	5.7	<0.0001	0.38	4.3
W7	8/24/2009														
W7	8/25/2009														
W7	8/31/2009														
W7	10/7/2009								0.017						
W7	10/7/2009	0.001	11.9	0.0346	0.0013	7.38	0.002	0.0015	<0.05	<0.0002	<0.0006	6.46	<0.0001	0.312	4
W7	10/25/2009	<0.005	12.6	0.029	0.002	7.41	0.001	<0.0002		<0.0005	0.0001	7.24	<0.005	0.361	<3
W7	10/25/2009														
W7	4/19/2010	<0.001	7.53	0.0207	0.0008	5.17	0.001	0.0001	0.1	<0.0002	<0.0006	4.44	<0.0001	0.218	2.8
W7	4/27/2010	<0.005	2.83	0.007	<0.001	1.89	0.001	<0.0002		<0.0005	<0.0001	3.14	<0.005	0.073	<3
W7	4/29/2010	<0.005	3.01	0.01	<0.001	1.91	0.001	<0.0002		<0.0005	<0.0001	3.2	<0.005	0.077	<3
W7	5/1/2010	<0.005	3.62	0.03	<0.001	2.25	0.002	0.0003	0.052	<0.0005	<0.0001	4.34	<0.005	0.095	<3
W7	5/3/2010	<0.005	3.78	0.033	<0.001	2.55	0.002	0.0003		<0.0005	<0.0001	4.17	<0.005	0.108	<3
W7	5/5/2010	<0.005	4.7	0.03	<0.001	3.03	0.002	0.002	0.058	<0.0005	<0.0001	4.7	<0.005	0.12	<3
W7	5/7/2010	<0.005	4.7	0.039	<0.001	3.01	0.003	0.0003	0.049	<0.0005	<0.0001	4.5	<0.005	0.13	<3
W7	5/9/2010	<0.005	6.2	0.016	<0.001	3.97	0.002	0.0003	0.027	<0.0005	0.0001	4.24	<0.005	0.17	<3
W7	5/10/2010	<0.005	7.83	0.018	0.001	5.1	0.002	0.0003	0.03	<0.0005	0.0002	5.37	<0.005	0.22	<3
W7	5/12/2010			<.01											
W7	5/12/2010	<0.005	7.14	0.021	0.001	4.83	0.001	0.0002	0.022	<0.0005	0.0001	5.23	<0.005	0.206	<3
W7	5/14/2010	<0.005	6.29	0.042	<0.001	4.05	0.003	0.0002		<0.0005	0.0001	4.69	<0.005	0.177	<3
W7	5/16/2010	<0.005	6.52	0.037	<0.001	4.02	0.001	<0.0002	0.041	<0.0005	0.0001	4.45	<0.005	0.188	<3
W7	5/17/2010	<0.005	6.47	0.045	0.001	3.92	0.001	0.0002	0.031	<0.0005	0.0002	4.32	<0.005	0.193	<3
W7	5/19/2010	<0.005	9.58	0.019	0.001	6.7	<0.003	<0.0002	0.024	<0.0005	0.0002	5.4	<0.005	0.261	<3
W7	5/21/2010	<0.01	8	0.035	0.001	5	0.002	0.0002		<0.0005	<0.0008	5.41	<0.005	0.246	<60
W7	5/23/2010	<0.01	10	0.13	0.001	5	0.007	0.0013		<0.0005	<0.0008	8.91	<0.005	0.261	<60
W7	5/25/2010	<0.01	13	0.158	0.002	8	0.008	0.0018		<0.0005	<0.001	13.2	<0.005	0.279	<60
W7	5/27/2010	<0.01	17	0.391	0.002	8	0.03	0.0052		0.0007	<0.001	24.4	<0.005	0.316	<60
W7	5/29/2010	<0.01	13	0.248	0.002	6	0.013	0.0025		<0.0005	<0.0008	12.9	<0.005	0.314	<60
W7	5/31/2010	<0.01	13	0.226	0.002	7	0.015	0.0028		<0.0005	<0.0008	14.2	<0.005	0.307	<60
W7	6/2/2010	<0.01	13	0.148	0.002	8	0.009	0.0014		<0.0005	<0.0008	10.2	<0.005	0.337	<60
W7	6/4/2010	<0.01	11	0.081	0.002	7	0.006	0.001		<0.0005	<0.0008	8.48	<0.005	0.311	<60
W7	6/6/2010	<0.01	12	0.091	0.002	7	0.006	0.0012		<0.0005	<0.0008	9.38	<0.005	0.337	<60
W7	6/14/2010	<0.01	12	0.073	0.002	7	0.004	0.0006		<0.0005	<0.0008	7.65	<0.005	0.356	<60
W7	6/16/2010	<0.01	12	0.036	0.002	7	0.002	<0.0002		<0.0005	<0.0008	6.5	<0.005	0.342	<60
W7	6/18/2010	<0.01	11	0.042	0.002	7	0.002	0.0003		<0.0005	<0.0008	6.69	<0.005	0.323	<60
W7	6/20/2010	<0.01	11	0.059	0.002	7	0.004	0.0007		<0.0005	<0.0008	8.16	<0.005	0.308	<60
W7	6/22/2010	<0.01	11	0.26	0.001	6	0.016	0.0031		<0.0005	<0.0008	18.4	<0.005	0.298	<60
W7	6/24/2010	<0.01	12	0.067	0.001	8	0.006	0.0006		<0.0005	<0.0008	8.82	<0.005	0.285	<60
W7	6/26/2010	<0.01	10	0.032	0.001	6	0.002	0.0003		<0.0005	<0.0008	6.6	<0.005	0.277	<60
W7	6/28/2010	<0.01	9	0.021	0.001	6	0.002	0.0003		<0.0005	<0.0008	7.12	<0.005	0.312	<60
W7	7/5/2010	<0.01	9	0.078	0.001	6	0.005	0.0008		<0.0005	<0.0008	9.55	<0.005	0.263	<60
W7	8/5/2010	<0.01	10	0.029	0.001	6	0.003	<0.0002		<0.0005	<0.0008	6.81	<0.005	0.306	<60
W7	9/12/2010	<0.01	9	0.052	<0.001	5	0.002	<0.0002		<0.0005	<0.0008	7.25	<0.005	0.253	<60
W7	9/12/2010	<0.01	9	0.026	<0.001	5	0.002	<0.0002		<0.0005	<0.0008	7.18	<0.005	0.248	<60
W7	10/8/2010	<0.005	11.1	0.036	0.001	6.52	0.002	<0.0002		<0.0005	<0.0001	7.13	<0.005	0.316	<3
W7	4/27/2011	<0.005	3.72	0.03	<0.001	2.16	<0.001	<0.0002	0.067	<0.0005	<0.0001	2.96	<0.005	0.089	<3
W7	5/2/2011	<0.005	2.78	0.039	<0.001	1.51	0.001	<0.0002		<0.0005	<0.0001	2.73	<0.005	0.069	<3
W7	6/1/2011	<0.005	8.93	0.035	0.001	5.45	0.002	0.0002	0.043	<0.0005	0.0001	5.82	<0.005	0.25	<3
W7	6/16/2011	<0.005	10.2	0.097	0.001	5.94	0.008	0.0012	0.226	<0.0005	0.0002	10.3	<0.005	0.24	<3
W7	7/1/2011	<0.005	7.82	0.162	0.001	5.16	0.012	0.0021	0.267	<0.0005	0.0003	11.9	<0.005	0.179	<3

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Total Li (mg/L)	Total Mg (mg/L)	Total Mn (mg/L)	Total Mo (mg/L)	Total Na (mg/L)	Total Ni (mg/L)	Total Pb (mg/L)	Total P (mg/L)	Total Sb (mg/L)	Total Se (mg/L)	Total Si (mg/L)	Total Sn (mg/L)	Total Sr (mg/L)	Total S (mg/L)
W7	8/2/2011	0.005	8.29	0.304	<0.001	4.04	0.019	0.0035	0.412	<0.0005	0.0003	22.3	<0.005	0.161	<3
W7	8/16/2011	<0.005	8.1	0.081	<0.001	4.64	0.003	0.0004	0.058	<0.0005	0.0001	7	<0.005	0.209	<3
W7	8/19/2011	<0.005	8.05	0.078	<0.001	4.62	0.003	0.0002	0.043	<0.0005	0.0001	7	<0.005	0.201	<3
W7	9/12/2011	<0.005	10.6	0.083	0.001	6.09	0.002	<0.0002	0.033	<0.0005	0.0002	7.49	<0.005	0.247	<3
W7	10/6/2011	<0.005	10.2	0.073	0.001	5.59	0.001	<0.0002	0.023	<0.0005	0.0001	6.27	<0.005	0.264	<3
W7	11/22/2011	<0.005	12.8	0.093	0.001	6.69	0.002	0.0008	0.077	<0.0005	0.0002	7.33	<0.005	0.358	6.5
W7	1/24/2012	<0.005	14.5	0.06	0.001	8.39	<0.001	<0.0002	0.014	<0.0005	0.0003	6.36	<0.005	0.449	5.9
W7	2/25/2012	<0.005	13.9	0.004	0.001	8.08	<0.001	<0.0002	0.012	<0.0005	0.0003	6.3	<0.005	0.468	6
W7	3/5/2012	<0.005	14.6	0.005	0.001	8.36	<0.001	<0.0002	0.016	<0.0005	0.0004	6.35	<0.005	0.485	5.6
W7	4/24/2012	<0.0050	3.17	0.0582	<0.0010	1.84	0.0019	0.00029	0.048	<0.00050	<0.00010	4.08	<0.0050	0.0798	<3.0
W7	5/6/2012	<0.0050	4.06	0.0133	<0.0010	2.47	0.0013	<0.00020	0.031	<0.00050	0.00014	3.65	<0.0050	0.105	<3.0
W7	5/6/2012	<0.0050	4.12	0.0145	<0.0010	2.53	0.0014	<0.00020	0.034	<0.00050	0.00011	3.57	<0.0050	0.103	<3.0
W7	6/19/2012	<0.0050	9.46	0.199	0.0012	5.45	0.0107	0.00178	0.222	<0.00050	0.0002	16.3	<0.0050	0.236	<3.0
W7	7/17/2012	<0.0050	11	0.0937	0.0015	6.4	0.0032	0.00026	0.074	<0.00050	0.00019	8.31	<0.0050	0.305	<3.0
W7	7/30/2012	<0.0050	11.6	0.0683	0.0014	6.83	0.0025	<0.00020	0.044	<0.00050	0.00022	6.59	<0.0050	0.318	3.3
W7	8/8/2012	<0.0050	12.7	0.0626	0.0014	7.25	0.0016	<0.00020	0.036	<0.00050	0.00028	7.38	<0.0050	0.334	<3.0
W7	8/23/2012	<0.0050	13	0.0627	0.0016	7.39	0.0016	<0.00020	0.035	<0.00050	0.00017	7.8	<0.0050	0.374	<3.0
W7	9/13/2012	<0.0050	10.5	0.126	<0.0010	6	0.0022	<0.00020	0.034	<0.00050	0.00021	6.83	<0.0050	0.269	<3.0
W7	10/16/2012	<0.0050	13	0.101	0.0012	7.25	0.0023	<0.00020	0.046	<0.00050	0.0002	7.29	<0.0050	0.327	<3.0
W7	11/2/2012	<0.0050	14.4	0.103	0.0013	7.87	0.0014	<0.00020	0.023	<0.00050	0.00023	6.3	<0.0050	0.411	5.1
W7	12/29/2012	<0.0050	19.9	0.0405	0.0012	11.7	0.0016	<0.00020	0.022	<0.00050	0.00032	7.67	<0.0050	0.488	7.9
W7	4/30/2013	<0.0050	19	0.053	0.0013	10.1	<0.0010	<0.00020	<0.01	<0.00050	0.00042	7.29	<0.0050	0.603	8.1
W7	5/14/2013	<0.0050	3	0.164	<0.0010	1.5	0.0029	0.00046	0.116	<0.00050	0.00011	4.78	<0.0050	0.0703	<3.0
W7	6/19/2013	<0.0050	9.98	0.0448	0.0013	6.12	0.0018	<0.00020	0.036	<0.00050	0.00018	6.43	<0.0050	0.275	<3.0
W7	6/19/2013	<0.0050	9.92	0.042	0.0014	6.28	0.0019	<0.00020	0.028	<0.00050	0.00012	6.47	<0.0050	0.275	<3.0
W7	7/13/2013	<0.0050	12.1	0.0336	0.0016	7.03	0.0017	<0.00020	0.027	<0.00050	0.0002	6.88	<0.0050	0.341	<3.0
W7	8/21/2013	<0.0050	13.3	0.0492	0.0016	7.75	0.0014	<0.00020	0.028	<0.00050	0.00014	6.97	<0.0050	0.364	<3.0
W7	9/6/2013	<0.0050	13.9	0.102	0.0013	8.11	0.0019	<0.00020	0.044	<0.00050	0.00017	7.96	<0.0050	0.364	<3.0
W7	10/21/2013	<0.0050	11.3	0.0934	0.0011	6.34	0.0016	<0.00020	0.027	<0.00050	0.00021	6.38	<0.0050	0.303	<3.0
W7	11/16/2013	<0.0050	15.6	0.0906	0.0017	8.83	<0.0010	<0.00020	0.026	<0.00050	0.00017	6.95	<0.0050	0.436	<3.0
W7	12/7/2013	<0.0050	17.5	0.162	0.0013	10.1	0.001	<0.00020	0.021	<0.00050	0.00019	6.56	<0.0050	0.476	5.8
C4	5/6/2012	0.0188	14.4	0.936	0.0015	3.45	0.0498	0.0126	1.11	0.00084	0.00069	57.5	<0.0050	0.168	<3.0
C4	8/1/2012	<0.0050	11.5	0.868	<0.0010	5.22	0.0139	0.00303	0.41	<0.00050	0.00019	18.8	<0.0050	0.237	<3.0
C4	8/8/2012	0.0057	12.3	1	0.001	5.55	0.0151	0.00371	0.478	<0.00050	0.00021	18	<0.0050	0.26	<3.0
C4	8/25/2012	<0.0050	14.3	1.32	<0.0010	5.64	0.0179	0.00439	0.657	<0.00050	0.00016	13	<0.0050	0.321	<3.0
C4	6/22/2013	0.0152	20.2	1.24	0.0023	7.08	0.0483	0.00977	1.28	0.00084	0.00059	40.7	<0.0050	0.337	3.6
C4	7/19/2013	0.0106	15.8	0.913	0.002	6.66	0.0314	0.00631	0.591	0.00064	0.00071	32	<0.0050	0.323	<3.0
C4	8/22/2013	0.0094	17.3	1.25	0.0014	6.83	0.0352	0.00784	1.01	0.00058	0.00053	31	<0.0050	0.346	<3.0
C4	9/6/2013	<0.0050	14.2	0.732	0.0014	6.74	0.0124	0.00208	0.342	<0.00050	0.00016	18.3	<0.0050	0.276	<3.0
C4	10/18/2013	<0.0050	11.8	0.427	<0.0010	6.15	0.0031	<0.00020	0.06	<0.00050	<0.00010	8.27	<0.0050	0.237	4.6
C4	10/18/2013	<0.0050	11.3	0.425	<0.0010	6.01	0.0029	<0.00020	0.065	<0.00050	0.00011	8.14	<0.0050	0.236	3.6
C4	11/17/2013	<0.0050	21.5	3.19	0.0011	8.09	0.0077	<0.00020	0.158	<0.00050	0.00027	9.05	<0.0050	0.409	8.2
C10	5/12/2012	0.0292	21.1	2.09	0.0021	4.3	0.074	0.0214	2.08	0.00117	0.00116	66.7	<0.0050	0.307	<3.0
C10	7/30/2012	0.015	16.2	0.947	0.0021	6.41	0.041	0.00995	0.967	0.00096	0.00073	42.7	<0.0050	0.315	<3.0
C10	7/30/2012														
C10	8/10/2012	0.0172	18.8	1.31	0.0017	6.4	0.0467	0.0125	1.15	0.00087	0.00057	44.2	<0.0050	0.363	<3.0
C10	8/27/2012	0.0198	20.2	1.7	0.0023	6.34	0.0524	0.013	1.39	0.00115	0.00086	44	<0.0050	0.404	<3.0
C10	6/22/2013	<0.0050	9.68	0.0986	0.0012	6.32	0.0044	0.0006	0.083	<0.00050	0.0001	8.47	<0.0050	0.223	<3.0
C10	7/23/2013	<0.0050	10.6	0.407	0.0012	6.52	0.0099	0.00212	0.239	<0.00050	0.00024	15.9	<0.0050	0.226	<3.0
C10	8/22/2013	<0.0050	10.7	0.11	<0.0010	6.49	0.0038	0.00044	0.073	<0.00050	0.00013	8.98	<0.0050	0.259	<3.0
C10	9/8/2013	<0.0050	10.2	0.172	0.0014	6.24	0.0033	0.00025	0.063	<0.00050	0.00017	7.06	<0.0050	0.256	<3.0
C10	9/8/2013	<0.0050	10.3	0.174	<0.0010	6.4	0.0032	0.00025	0.067	<0.00050	0.00014	7.27	<0.0050	0.255	<3.0
C10	10/18/2013	<0.0050	10.4	0.321	<0.0010	6.3	0.0038	0.00054	0.105	<0.00050	0.00015	8.99	<0.0050	0.231	<3.0

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Total Te (mg/L)	Total Th (mg/L)	Total Ti (mg/L)	Total Tl (mg/L)	Total U (mg/L)	Total V (mg/L)	Total W (mg/L)	Total Zn (mg/L)	Total Zr (mg/L)	Total N (mg/L)	Ammonia (mg/L)	Ammonium (mg/L)	Ammonium (mg/L)	NO2+NO3 (mg/L)
W1	5/27/2005			0.022	<0.00020	0.0006	<0.030		0.0096			0.039			
W1	6/30/2005			<0.010	<0.00010	0.00113	<0.0010		<0.0010			<0.020			
W1	7/28/2005			<0.010	<0.00010	0.00122	<0.0010		<0.0010			<0.020			
W1	8/29/2005			<0.010	<0.00020	0.00077	<0.030		<0.0050			0.041			
W1	9/28/2005			<0.010	<0.00020	0.00058	<0.030		<0.0050			0.026			
W1	10/15/2005			<0.010	<0.00010	0.000786	0.0013		0.001			0.02			
W2	5/27/2005			0.016	<0.00020	0.00054	<0.030		<0.0050			0.026			
W2	5/27/2005			0.057		0.00096			0.006			<0.02			
W2	6/30/2005			<0.010	<0.00010	0.00108	<0.0010		<0.0010			<0.020			
W2	7/28/2005			<0.010	<0.00010	0.00136	<0.0010		<0.0010			<0.020			
W2	8/30/2005			<0.010	<0.00020	0.00082	<0.030		<0.0050			0.029			
W2	9/28/2005			<0.010	<0.00020	0.00059	<0.030		<0.0050			<0.020			
W2	10/15/2005			<0.010	<0.00010	0.000756	0.001		0.0011			<0.020			
W3	5/27/2005			<0.010	<0.00020	0.00035	<0.030		<0.0050			<0.020			
W3	6/30/2005			<0.010	<0.00010	0.0009	<0.0010		<0.0020			<0.020			
W3	7/29/2005			0.045	<0.00010	0.000778	0.0032		<0.0070			<0.020			
W3	8/29/2005			<0.010	<0.00020	0.00051	<0.030		<0.0050			0.042			
W3	9/28/2005			<0.010	<0.00020	0.00039	<0.030		<0.0050			<0.020			
W3	10/15/2005			<0.010	<0.00010	0.00044	<0.0010		<0.0010			0.023			
W8	5/27/2005			<0.010	<0.00020	<0.00020	<0.030		<0.0050			<0.020			
W8	6/30/2005			<0.010	<0.00010	0.000063	<0.0010		<0.0020			0.041			
W8	7/29/2005			<0.010	<0.00010	0.000083	<0.0010		0.0011			<0.020			
W8	8/30/2005			<0.010	<0.00020	<0.00020	<0.030		<0.0050			0.023			
W8	9/28/2005			<0.010	<0.00020	<0.00020	<0.030		<0.0050			<0.020			
W8	10/15/2005			<0.010	<0.00010	0.0001	<0.0010		0.0011			0.026			
W9	5/27/2005			<0.010	<0.00020	<0.00020	<0.030		<0.0050			<0.020			
W9	6/30/2005			<0.010	<0.00010	0.000027	<0.0010		<0.0050			0.02			
W9	7/29/2005			<0.010	<0.00010	0.000052	<0.0010		<0.0040			<0.020			
W9	8/29/2005			<0.010	<0.00020	<0.00020	<0.030		<0.0050			0.026			
W9	9/28/2005			<0.010	<0.00020	<0.00020	<0.030		<0.0050			<0.020			
W9	10/15/2005			<0.010	<0.00010	0.000025	<0.0010		<0.0010			<0.020			
W10	5/27/2005			<0.010	<0.00020	<0.00020	<0.030		<0.0050			0.027			
W10	5/27/2005			<0.01	<0.0002	0.00092	<0.03		<0.005			<0.02			
W10	6/30/2005			<0.010	<0.00010	0.000034	0.0015		0.0289			0.036			
W10	7/29/2005			0.09	<0.00010	0.000097	0.0067		0.017			<0.020			
W10	8/29/2005			0.018	<0.00020	<0.00020	<0.030		0.0071			0.033			
W10	9/28/2005			<0.010	<0.00020	<0.00020	<0.030		<0.0050			<0.020			
W10	10/15/2005			<0.010	<0.00010	0.000017	<0.0010		0.0014			<0.020			
W6	5/27/2005			<0.010	<0.00020	<0.00020	<0.030		<0.0050			<0.020			
W6	6/30/2005			<0.010	<0.00010	0.000151	0.0012		<0.0030			<0.020			
W6	7/29/2005			0.033	<0.00010	0.00106	0.0027		<0.0060			<0.020			
W6	8/30/2005			<0.010	<0.00020	<0.00020	<0.030		<0.0050			<0.020			
W6	9/28/2005			<0.010	<0.00020	<0.00020	<0.030		<0.0050			<0.020			
W6	10/15/2005			<0.010	<0.00010	0.000129	<0.0010		<0.0010			<0.020			
W6	6/2/2006			0.077	<0.0001	0.000215	0.0066		0.0085			0.021			
W6	8/25/2006			<0.0010	<0.0001	<0.001	0.0008		0.005	<0.002					
W6	9/6/2006			0.0014	<0.00005	<0.0005	0.0008		<0.001	<0.001					
W6	6/20/2007			<0.0005	<0.00005	<0.0005	0.0011		0.004	<0.001		<0.05			<0.1
W6	9/18/2007			0.0015	<0.00005	<0.0005	0.0007		0.008	<0.001		<0.05			<0.1
W6	6/17/2008	<0.0001	<0.0001	0.0059	<0.00001	<0.0004	0.00094		0.007	0.0004		<0.05			0.01
W6	9/18/2008	<0.0001	<0.0001	0.0015	<0.00001	<0.0004	0.00065		0.004	0.0003		<0.05			<0.01
W6	4/14/2009	0.0001	0.0001	0.064	<0.00001	0.0041	0.00381		0.276	0.0038			0.12	0.12	0.59
W6	4/30/2009	<0.0001	<0.0001	0.222	0.00003	0.0004	0.0142		0.023	0.0013			<0.05	<0.05	0.04
W6	7/28/2009														
W6	8/6/2009	<0.0001	<0.0004	0.0024	<0.00001	0.0015	0.0008		0.003	<0.0001					
W6	8/6/2009	<0.0001	<0.0004	0.0019	<0.00001	0.0015	0.0007		0.006	0.0001			0.06	0.06	2.93
W6	8/18/2009														
W6	8/25/2009														
W6	10/7/2009														
W6	10/7/2009	<0.0001	<0.0004	0.0022	<0.00001	0.0005	0.0006		0.005	<0.0001			0.1	0.1	8.18
W6	10/25/2009			<0.005	<0.00005	0.0009	<0.005		<0.005	<0.0005		0.134			7.9
W6	4/19/2010	<0.0001	<0.0004		<0.00001	0.0014	0.0043		0.014	0.0003			<0.05	<0.05	0.54
W6	5/16/2010			<0.005	<0.00005	0.0001	<0.005	<0.001	<0.005	<0.0005		0.046			<0.02
W6	5/17/2010			<0.005	<0.00005	0.0004	<0.005	<0.001	<0.005	<0.0005		<0.005			<0.02
W6	5/19/2010			<0.005	<0.00005	0.0005	<0.005	<0.001	<0.005	<0.0005		0.08			<0.02
W6	5/21/2010			<0.01	<0.00005	0.0006	<0.005		<0.01	<0.002		0.006			<0.02
W6	5/23/2010			<0.01	<0.00005	0.0006	<0.005		<0.01	<0.002		0.031			<0.02
W6	5/25/2010			<0.01	<0.00005	0.0006	<0.005		<0.01	<0.002		0.17			<0.02
W6	5/29/2010			<0.01	<0.00005	0.0005	<0.005		<0.01	<0.002		0.08			<0.02
W6	6/2/2010			<0.01	<0.00005	0.003	<0.005		<0.01	<0.002		0.031			0.97
W6	6/4/2010			<0.01	<0.00005	0.0003	<0.005		<0.01	<0.002		0.034			<0.02
W6	6/6/2010			<0.01	<0.00005	0.0004	<0.005		<0.01	<0.002		0.06			<0.02
W6	6/14/2010			0.011	<0.00005	0.0003	<0.005		<0.01	<0.002		0.016			<0.02
W6	6/16/2010			0.012	<0.00005	0.0003	<0.005		<0.01	<0.002		0.013			0.03
W6	6/18/2010			<0.01	<0.00005	0.0003	<0.005		<0.01	<0.002		<0.5			<0.02
W6	6/20/2010			<0.01	<0.00005	0.0003	<0.005		<0.01	<0.002		<0.5			0.02
W6	6/22/2010			<0.01	<0.00005	0.0003	<0.005		<0.01	<0.002		0.01			<0.02
W6	6/24/2010			<0.01	<0.00005	0.0002	<0.005		<0.01	<0.002		0.012			<0.02
W6	6/26/2010			<0.01	<0.00005	0.0003	<0.005		<0.01	<0.002		0.034			<0.02
W6	6/26/2010														
W6	6/28/2010			0.019	<0.00005	0.0003	<0.005		<0.01	<0.002		0.031			<0.02
W6	7/5/2010			0.051	<0.00005	0.0004	<0.005		<0.01	<0.002		0.15			<0.02
W6	8/5/2010			<0.01	<0.00005	0.0002	<0.005		<0.01	<0.002		0.014			<0.02
W6	9/9/2010			0.018	<0.00005	0.0003	<0.005		0.015	<0.002		<0.005			<0.02
W6	10/8/2010			0.016	<0.00005	0.0003	<0.005		<0.005	<0.0005		0.053			<0.02
W6	8/19/2011			<0.005	<0.00005	0.0025	<0.005		0.008	<0.0005		0.006			1.05
W6	7/17/2012			0.747	0.000147	0.0022	0.0672		0.073	0.00302		0.14			0.026
W6	7/30/2012			0.529	0.000108	0.0017	0.0475		0.0504	0.00202		0.07			0.059
W6	8/9/2012			0.299	0.000068	0.00102	0.0252		0.027	0.00116		0.025			0.029
W6	8/25/2012			0.063	<0.000050	0.00167	0.0348		0.0392	0.00101		0.023			0.045
W6	9/13/2012			0.0071	<0.000050	0.00018	<0.0050		<0.0050	<0.00050		0.011			0.034
W6	10/13/2012			0.0098	<0.000050	0.00027	<0.0050		<0.0050	<0.00050		0.016			0.042
W6	11/12/2012			<0.0050	<0.000050	0.00034	<0.0050		<0.0050	<0.00050		0.025			<0.020
W6	5/14/2013			0.024	<0.000050	<0.00010	<0.0050		0.0057	<0.00050		0.13			<0.020
W6	6/22/2013			0.2	0.000066	0.0005	0.014		0.0169	0.0016		0.024			0.042
W6	7/12/2013			0.0218	<0.000050	0.00022	<0.0050		<0.0050	<0.00050		0.012			<0.020
W6	8/22/2013			<0.0050	<0.000050	0.00017	<0.0050		<0.0050	<0.00050		0.014			<0.020
W6	9/6/2013			0.007	<0.000050	0.0002	<0.0050		<0.0050	<0.00050		0.0061			0.03
W6	10/7/2013			0.0916	<0.000050	0.0003	0.0074		0.0079	0.0006		0.044			0.042
W6	11/16/2013			0.0133	<0.000050	0.00022	<0.0050		<0.0050	<0.00050		0.015			<0.020
W6	12/7/2013			<0.0050	<0.000050	0.00046	<0.0050		<0.0050	<0.00050		0.0076			<0.020
W7	5/27/2005			0.022	<0.00020	0.00047	<0.030		<0.0050			<0.020			
W7	6/30/2005			<0.010	<0.00010	0.000769	0.0013		<0.0020			<0.020			
W7	7/29/2005			0.027	<0.00010	0.0012	0.0024		<0.0050			<0.020			
W7	8/30/2005			<0.010	<0.00020	0.00058	<0.030		<0.0050			0.02			
W7	9/28/2005			0.06	<0.00020	0.00079	<0.030		0.0101			<0.020			
W7	10/15/2005			<0.010	<0.00010	0.000547	<0.0010		0.0017			0.021			
W7	6/2/2006			0.378	<0.0001	0.00201	0.0379		0.0484			0.062			
W7	6/8/2006			0.015	<0.0001	0.000529	0.0035		0.0048						

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Total Te (mg/L)	Total Th (mg/L)	Total Ti (mg/L)	Total Tl (mg/L)	Total U (mg/L)	Total V (mg/L)	Total W (mg/L)	Total Zn (mg/L)	Total Zr (mg/L)	Total N (mg/L)	Ammonia (mg/L)	Ammonium (mg/L)	Ammonium (mg/L)	NO2+NO3 (mg/L)
W7	6/15/2006											0.027			
W7	6/15/2006			0.0685	<0.00005	0.0008	0.0062		0.006	0.001					
W7	6/23/2006			0.257	<0.00005	0.0014	0.0173		0.022	0.002		0.028			
W7	6/28/2006			0.0822	<0.00005	0.0011	0.0064		0.006	0.001		0.01			
W7	7/7/2006			0.0044	<0.00005	0.0011	0.001		<0.001	<0.001					
W7	7/12/2006			0.0036	<0.00005	0.001	0.0009		0.015	<0.001		0.014			
W7	7/20/2006			0.0026	<0.00005	0.0012	0.0008		0.002	<0.001		0.012			
W7	7/26/2006											0.016			
W7	7/26/2006			0.006	<0.00005	0.0012	0.0011		0.002	<0.001					
W7	8/2/2006			0.004	<0.00005	0.001	0.001		0.001	<0.001		0.015			
W7	8/10/2006			0.0018	<0.00005	0.0009	0.0009		0.002	<0.001		0.014			
W7	8/25/2006											0.027			
W7	8/25/2006			<0.0010	<0.0001	0.001	0.0009		0.004	<0.002					
W7	8/30/2006			0.003	<0.00005	0.0011	0.0008		0.003	<0.001		0.012			
W7	9/6/2006			0.0026	<0.00005	0.0011	0.0009		0.001	<0.001		0.011	0.011	0.011	
W7	9/13/2006			0.0014	<0.00005	0.0006	0.0005		<0.001	<0.001		0.011			
W7	9/20/2006			0.0037	<0.00005	0.0011	0.0009		0.005	<0.001		0.008			
W7	9/28/2006			0.002	<0.0001	0.001	0.0009		0.008	<0.002		0.011			
W7	10/4/2006			0.0042	<0.0001	0.001	0.001		<0.002	<0.002		0.079			
W7	10/12/2006			0.0124	<0.00005	0.0011	0.0014		0.006	<0.001		0.024			
W7	6/5/2007			0.0776	<0.00005	0.0014	0.0054		0.009	0.001		<0.05			<0.1
W7	6/5/2007			0.0898	<0.0001	0.001	0.0059		0.01	<0.002		<0.05			0.28
W7	6/5/2007			0.0915	<0.00005	0.0015	0.006		0.012	0.001		<0.05			<0.1
W7	6/20/2007			0.0055	<0.00005	0.0014	0.0016		0.004	<0.001		<0.05			0.18
W7	7/18/2007			0.0096	<0.00005	0.0012	0.0014		0.005	<0.001		<0.05			0.22
W7	7/18/2007			0.0103	<0.00005	0.0013	0.0014		0.006	<0.001		<0.05			0.2
W7	7/18/2007			0.0096	<0.00005	0.0012	0.0013		0.003	<0.001		<0.05			0.2
W7	8/24/2007											<0.05			
W7	8/24/2007											<0.05			
W7	8/24/2007											<0.05			
W7	8/24/2007			<0.0005	<0.00005	0.0012	0.0013		0.01	<0.001					0.14
W7	8/24/2007			<0.0005	<0.00005	0.0012	0.0011		0.009	<0.001					0.14
W7	8/24/2007			<0.0005	<0.00005	0.0013	0.0012		0.006	<0.001					0.16
W7	8/24/2007			<0.0005	<0.00005	0.0012	0.0011		0.009	<0.001					0.14
W7	8/24/2007			<0.0005	<0.00005	0.0013	0.0012		0.006	<0.001					0.16
W7	9/18/2007			0.016	<0.00005	0.0008	0.002		0.008	<0.001		<0.05			0.06
W7	10/30/2007			0.0066	<0.00005	0.0009	0.0011		0.013	<0.001		<0.05			0.2
W7	10/30/2007			0.0043	<0.00005	0.0009	0.0008		0.011	<0.001		<0.05			0.24
W7	10/30/2007			0.0041	<0.00005	0.0009	0.0008		0.01	<0.001		<0.05			0.19
W7	10/30/2007														
W7	4/22/2008	<0.0001	<0.0001	0.0004	<0.00001	0.0021	0.00032		0.004	<0.0001		<0.05			<0.01
W7	4/22/2008	0.0001	0.0002	0.0017	0.00002	0.0021	0.0003		0.005	0.0001		<0.05			<0.01
W7	4/22/2008	<0.0001	0.0002	0.0003	0.00001	0.0023	0.0003		0.007	0.0002		<0.05			<0.01
W7	4/22/2008														
W7	6/3/2008	<0.0001	<0.0001	0.0128	<0.00001	0.0011	0.00207		0.004	0.0004		<0.05			0.03
W7	6/3/2008	<0.0001	<0.0001	0.0122	<0.00001	0.0011	0.002		0.005	0.0005		<0.05			0.04
W7	6/3/2008	<0.0001	<0.0001	0.0134	<0.00001	0.0011	0.00224		0.007	0.0005		<0.05			0.04
W7	6/3/2008														
W7	6/17/2008	<0.0001	<0.0001	0.0081	<0.00001	0.0014	0.0014		0.007	0.0004		<0.05			0.14
W7	8/6/2008			0.0357	<0.00005	0.0012	0.0031		0.01	<0.001		<0.05			0.16
W7	8/6/2008			0.0618	<0.00005	0.0011	0.004		0.011	<0.001		<0.05			0.16
W7	8/6/2008			0.0628	<0.00005	0.0012	0.0042		0.01	<0.001					
W7	8/6/2008											<0.05			0.14
W7	9/18/2008	<0.0001	<0.0001	0.0194	<0.00001	0.0004	0.00224		0.009	0.0006		<0.05			0.04
W7	10/28/2008	<0.0001	<0.0001	0.0045	<0.00001	0.0007	0.00086		0.008	0.0003		<0.05			0.13
W7	10/28/2008	<0.0001	<0.0001	0.0026	<0.00001	0.0007	0.00073		0.121	0.0003		<0.05			0.12
W7	10/28/2008	<0.0001	<0.0001	0.002	<0.00001	0.0007	0.00072		0.049	0.0003		<0.05			0.13
W7	10/28/2008														
W7	4/30/2009	<0.0001	<0.0001	0.0046	<0.00001	<0.0004	0.00037		0.008	0.0002			<0.05	<0.05	<0.01
W7	6/8/2009	<0.0001	<0.0001	0.0256	<0.00001	0.001	0.00265		0.002	0.0005			<0.05	<0.05	0.15
W7	7/28/2009														
W7	8/6/2009	<0.0001	<0.0004	0.0034	<0.00001	0.0014	0.0009		0.003	0.0002			<0.05	<0.05	0.18
W7	8/24/2009														
W7	8/25/2009														
W7	8/31/2009														
W7	10/7/2009												<0.01	<0.01	
W7	10/7/2009	<0.0001	<0.0004	0.0068	<0.00001	0.0009	0.001		0.008	0.0002			<0.05	<0.05	0.12
W7	10/25/2009			<0.005	<0.00005	0.0013	<0.005		<0.005	<0.0005		0.01			0.15
W7	10/25/2009														
W7	4/19/2010	<0.0001	<0.0004		<0.00001	0.0005	0.0005		0.008	0.0002			0.16	0.16	0.01
W7	4/27/2010			<0.005	<0.00005	0.0001	<0.005		<0.005	<0.0005		0.015			<0.02
W7	4/29/2010			<0.005	<0.00005	0.0001	<0.005		<0.005	<0.0005		0.016			<0.04
W7	5/1/2010			0.015	<0.00005	0.0002	<0.005	<0.001	<0.005	<0.0005		0.031			0.08
W7	5/3/2010			0.014	<0.00005	0.0002	<0.005		<0.005	<0.0005		0.016			<0.02
W7	5/5/2010			0.016	<0.00005	0.0002	<0.005	<0.001	0.007	<0.0005		<0.01			<0.02
W7	5/7/2010			0.016	<0.00005	0.0003	<0.005	<0.001	<0.005	<0.0005		0.051			0.11
W7	5/9/2010			<0.005	<0.00005	0.0005	<0.005	<0.001	<0.005	<0.0005		<0.005			0.07
W7	5/10/2010			<0.005	<0.00005	0.0007	<0.005	<0.001	0.005	<0.0005		0.015			0.14
W7	5/12/2010														
W7	5/12/2010			<0.005	<0.00005	0.0006	<0.005	<0.001	0.007	<0.0005		<0.05			0.03
W7	5/14/2010			0.008	<0.00005	0.0005	<0.005		<0.005	<0.0005		0.034			<0.02
W7	5/16/2010			0.008	<0.00005	0.0007	<0.005	<0.001	<0.005	<0.0005		0.068			0.03
W7	5/17/2010			0.01	<0.00005	0.0007	<0.005	<0.001	<0.005	<0.0005		0.037			<0.02
W7	5/19/2010			0.007	<0.00005	0.0008	<0.005	<0.001	<0.005	<0.0005		0.02			0.06
W7	5/21/2010			0.016	<0.00005	0.0009	<0.005		<0.01	<0.002		0.017			0.05
W7	5/23/2010			0.086	<0.00005	0.0012	0.007		0.012	<0.002		0.013			0.07
W7	5/25/2010			0.143	0.00005	0.0011	0.01		0.017	<0.002		0.07			0.1
W7	5/27/2010			0.285	0.00017	0.0016	0.024		0.049	<0.002		0.052			0.13
W7	5/29/2010			0.117	0.00006	0.0017	0.012		0.021	<0.002		0.1			0.23
W7	5/31/2010			0.14	0.00007	0.002	0.013		0.023	<0.002		<0.05			0.14
W7	6/2/2010			0.105	<0.00005	0.0018	0.01		0.011	<0.002		<0.01			0.17
W7	6/4/2010			0.057	<0.00005	0.0015	0.005		<0.01	<0.002		0.34			0.18
W7	6/6/2010			0.082	<0.00005	0.0018	0.006		0.011	<0.002		0.1			0.18
W7	6/14/2010			0.051	<0.00005	0.0018	<0.005		<0.01	<0.002		0.016			0.16
W7	6/16/2010			0.016	<0.00005	0.0016	<0.005		<0.01	<0.002		0.016			0.17
W7	6/18/2010			0.02	<0.00005	0.0015	<0.005		<0.01	<0.002			0.052		0.15
W7	6/20/2010			0.043	<0.00005	0.0014	<0.005		<0.01	<0.002					0.13
W7	6/22/2010			0.174	0.00007	0.0012	0.017		0.027	<0.002		0.014			0.12
W7	6/24/2010			0.049	<0.00005	0.0009	0.005		<0.01	<0.002		0.012			0.08
W7	6/26/2010			0.019	<0.00005	0.0009	<0.005		<0.01	<0.002		0.009			0.1
W7	6/28/2010			0.01	<0.00005	0.001	<0.005		<0.01	<0.002		0.022			0.13
W7	7/5/2010			0.061	<0.00005	0.0007	<0.005		<0.01	<0.002		0.027			0.07
W7	8/5/2010			<0.01	<0.00005	0.0008	<0.005		<0.01	<0.002					
W7	9/12/2010			0.014	<0.00005	0.0005	<0.005		<0.01	<0.002					
W7	9/12/2010			0.012	<0.00005	0.0005	<0.005		<0.01	<0.002		0.039			0.03
W7	10/8/2010			<0.005	<0.00005	0.0008	<0.005		<0.005	<0.0005		0.069			0.07
W7	4/27/2011			<0.005	<0.00005	0.0002	<0.005		<0.005	<0.0005		0.01			<0.02
W7	5/2/2011			0.009	<0.00005	0.0001	<0.005		0.019	<0.0005					<0.02
W7	6/1/2011			0.022	<0.00005	0.0007	<0.005		<0.005	<0.0005		0.012			
W7	6/16/2011			0.114	<0.00005	0.0007	0.008		0.013	0.0007		0.013			0.1
W7	7/1/2011			0.126	<0.00005	0.0009	0.015		0.016	0.0012		0.016			0.05

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Total Te (mg/L)	Total Th (mg/L)	Total Ti (mg/L)	Total Tl (mg/L)	Total U (mg/L)	Total V (mg/L)	Total W (mg/L)	Total Zn (mg/L)	Total Zr (mg/L)	Total N (mg/L)	Ammonia (mg/L)	Ammonium (mg/L)	Ammonium (mg/L)	NO2+NO3 (mg/L)
W7	8/2/2011			0.354	0.00008	0.0009	0.031		0.029	0.002					0.06
W7	8/16/2011			0.016	<0.00005	0.0005	<0.005		0.017	0.0005		0.015			0.06
W7	8/19/2011			0.02	<0.00005	0.0004	<0.005		<0.005	<0.0005		0.006			0.04
W7	9/12/2011			<0.005	<0.00005	0.0005	<0.005		<0.005	<0.0005		0.025			0.05
W7	10/6/2011			<0.005	<0.00005	0.0006	<0.005		<0.005	<0.0005		0.019			0.11
W7	11/22/2011			0.02	<0.00005	0.001	<0.005		0.009	<0.0005		0.0214			0.136
W7	1/24/2012			<0.005	<0.00005	0.0019	<0.005		<0.005	<0.0005		0.005			0.324
W7	2/25/2012			<0.005	<0.00005	0.0015	<0.005		<0.005	<0.0005		0.006			0.289
W7	3/5/2012			<0.005	<0.00005	0.0015	<0.005		<0.005	<0.0005		0.0067			0.195
W7	4/24/2012			0.0264	<0.000050	0.00018	<0.0050		<0.0050	<0.00050		<0.0050			0.02
W7	5/6/2012			0.0055	<0.000050	<0.00010	<0.0050		<0.0050	<0.00050		0.01			<0.020
W7	5/6/2012			0.0062	<0.000050	<0.00010	<0.0050		<0.0050	<0.00050		0.011			<0.020
W7	6/19/2012			0.182	0.00005	0.00072	0.0164		0.015	0.00125		0.18			0.098
W7	7/17/2012			0.0379	<0.000050	0.00076	<0.0050		<0.0050	<0.00050		0.065			0.089
W7	7/30/2012			0.0092	<0.000050	0.00088	<0.0050		0.0063	<0.00050		0.13			0.124
W7	8/8/2012			<0.0050	<0.000050	0.00087	<0.0050		<0.0050	<0.00050		0.014			0.13
W7	8/23/2012			<0.0050	<0.000050	0.00097	<0.0050		<0.0050	<0.00050		0.049			0.21
W7	9/13/2012			<0.0050	<0.000050	0.00054	<0.0050		0.0055	<0.00050		0.0087			0.078
W7	10/16/2012			0.0061	<0.000050	0.00084	<0.0050		<0.0050	<0.00050		0.028			0.125
W7	11/2/2012			<0.0050	<0.000050	0.00137	<0.0050		0.0062	<0.00050		0.026			0.219
W7	12/29/2012			<0.0050	<0.000050	0.00121	<0.0050		<0.0050	<0.00050		0.025			0.063
W7	4/30/2013			<0.0050	<0.000050	0.00209	<0.0050		<0.0050	<0.00050		0.0082			0.132
W7	5/14/2013			0.0551	<0.000050	0.00016	<0.0050		0.0059	<0.00050		0.066			<0.020
W7	6/19/2013			0.0122	<0.000050	0.001	<0.0050		<0.0050	<0.00050		0.052			0.143
W7	6/19/2013			0.0094	<0.000050	0.00097	<0.0050		<0.0050	<0.00050		0.04			0.141
W7	7/13/2013			0.0073	<0.000050	0.001	<0.0050		<0.0050	<0.00050		0.023			0.17
W7	8/21/2013			0.0086	<0.000050	0.00114	<0.0050		<0.0050	<0.00050		0.051			0.157
W7	9/6/2013			0.0127	<0.000050	0.00095	<0.0050		<0.0050	<0.00050		0.017			0.128
W7	10/21/2013			<0.0050	<0.000050	0.00083	<0.0050		<0.0050	<0.00050		0.048			0.168
W7	11/16/2013			<0.0050	<0.000050	0.00148	<0.0050		<0.0050	<0.00050		0.027			0.257
W7	12/7/2013			<0.0050	<0.000050	0.00139	<0.0050		<0.0050	<0.00050		0.025			0.206
C4	5/6/2012			1.21	0.000244	0.00209	0.0946		0.0961	0.00346		0.088			0.029
C4	8/1/2012			0.248	0.000063	0.00116	0.0276		0.0272	0.0016		0.089			0.091
C4	8/8/2012			0.254	0.000066	0.00144	0.0292		0.0279	0.00138		0.09			0.107
C4	8/25/2012			0.08	<0.000050	0.00159	0.0377		0.0325	0.00202		0.005			<0.020
C4	6/22/2013			0.8	<0.000050	0.00305	0.0731		0.0872	0.00409		0.36			0.377
C4	7/19/2013			0.466	0.000122	0.00277	0.0477		0.0502	0.00309		0.26			0.089
C4	8/22/2013			0.497	0.000103	0.00228	0.055		0.0597	0.003		0.2			0.095
C4	9/6/2013			0.237	0.000053	0.00119	0.0188		0.0172	0.00149		0.12			0.07
C4	10/18/2013			0.0165	<0.000050	0.00083	<0.0050		<0.0050	<0.00050		0.091			0.069
C4	10/18/2013			0.017	<0.000050	0.00079	<0.0050		<0.0050	<0.00050		0.08			0.065
C4	11/17/2013			0.006	<0.000050	0.00063	<0.0050		<0.0050	0.001		0.28			<0.20
C10	5/12/2012			1.17	0.000369	0.00532	0.142		0.186	0.00514		0.18			0.058
C10	7/30/2012			0.571	0.000202	0.00449	0.0653		0.0956	0.0033		0.2			0.207
C10	7/30/2012														
C10	8/10/2012			0.554	0.000202	0.00407	0.0718		0.105	0.00332		0.12			0.196
C10	8/27/2012			0.805	0.000241	0.00375	0.0836		0.119	0.00453		<0.0050			0.103
C10	6/22/2013			0.0562	<0.000050	0.00114	0.0053		0.0055	0.00094		0.19			0.172
C10	7/23/2013			0.179	0.000067	0.00113	0.0156		0.0198	0.0017		0.14			<0.20
C10	8/22/2013			0.0342	<0.000050	0.00119	<0.0050		<0.0050	0.00065		0.056			0.143
C10	9/8/2013			0.017	<0.000050	0.00099	<0.0050		<0.0050	<0.00050		0.046			0.183
C10	9/8/2013			0.0166	<0.000050	0.00099	<0.0050		<0.0050	<0.00050		0.055			0.199
C10	10/18/2013			0.0427	<0.000050	0.00091	<0.0050		0.0054	0.00051		0.1			0.167

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	N-NO2 (mg/L)	N-NO3 (mg/L)	N-TKN (mg/L)	P-PO4-T (mg/L)	P-TDP (mg/L)	P-TRP (Ortho) (mg/L)	C-TIC (mg/L)	C-TOC (mg/L)	C-DOC (mg/L)	Cation Sum (meq/L)	Anion Sum (meq/L)	Hardness (CaCO3) (mg/L)	Acidity (pH 4.5) (mg/L)	Acidity pH 8.3 (mg/L)
W1	5/27/2005	0.002	0.0539												
W1	6/30/2005	<0.0010	0.0626												
W1	7/28/2005	<0.0010	0.0584												
W1	8/29/2005	<0.0010	0.0217												
W1	9/28/2005	<0.0010	0.0482												
W1	10/15/2005	<0.0010	0.132												
W2	5/27/2005	0.0015	0.0565												
W2	5/27/2005	<0.001	0.0339												
W2	6/30/2005	<0.0010	0.0621												
W2	7/28/2005	<0.0010	0.0124												
W2	8/30/2005	<0.0010	0.0141												
W2	9/28/2005	0.001	0.0458												
W2	10/15/2005	<0.0010	0.133												
W3	5/27/2005	<0.0010	0.0121												
W3	6/30/2005	<0.0010	0.0519												
W3	7/29/2005	<0.0010	0.0284												
W3	8/29/2005	<0.0010	<0.0050												
W3	9/28/2005	<0.0010	0.0132												
W3	10/15/2005	<0.0010	0.0393												
W8	5/27/2005	<0.0010	0.0225												
W8	6/30/2005	<0.0010	<0.0050												
W8	7/29/2005	<0.0010	<0.0050												
W8	8/30/2005	0.0011	<0.0050												
W8	9/28/2005	<0.0010	<0.0050												
W8	10/15/2005	<0.0010	0.0374												
W9	5/27/2005	<0.0010	<0.0050												
W9	6/30/2005	<0.0010	<0.0050												
W9	7/29/2005	<0.0010	<0.0050												
W9	8/29/2005	0.0016	<0.0050												
W9	9/28/2005	<0.0010	<0.0050												
W9	10/15/2005	<0.0010	<0.0050												
W10	5/27/2005	<0.0010	0.0094												
W10	5/27/2005	<0.001													
W10	6/30/2005	<0.0010	<0.0050												
W10	7/29/2005	<0.0010	<0.0050												
W10	8/29/2005	<0.0010	<0.0050												
W10	9/28/2005	<0.0010	<0.0050												
W10	10/15/2005	<0.0010	<0.0050												
W6	5/27/2005	0.0016	0.0316												
W6	6/30/2005	<0.0010	0.0086												
W6	7/29/2005	<0.0010	0.0223												
W6	8/30/2005	<0.0010	<0.0050												
W6	9/28/2005	<0.0010	<0.0050												
W6	10/15/2005	<0.0010	0.0061												
W6	6/2/2006	<0.001	0.0241												
W6	8/25/2006		<0.03												
W6	9/6/2006		<0.03												
W6	6/20/2007	<0.05	0.1	0.6	0.14	<0.02									
W6	9/18/2007	<0.02	<0.02	0.3	0.04	<0.02									
W6	6/17/2008	0.07	0.01	0.44	0.07										
W6	9/18/2008	<0.01	<0.01	0.47	<0.05		0.1		17.3	17.7					
W6	4/14/2009			5.73	0.06										
W6	4/30/2009			1.22	0.05										
W6	7/28/2009														
W6	8/6/2009														
W6	8/6/2009		2.92	0.5	0.02		<0.05								
W6	8/18/2009														
W6	8/25/2009								9.2	7.9					
W6	10/7/2009														
W6	10/7/2009		8.09	0.54	0.05				6.4	6.3					
W6	10/25/2009	0.061	7.8				<0.005								
W6	4/19/2010			0.53	0.05										
W6	5/16/2010	<0.005	<0.02				<0.005			5.3					
W6	5/17/2010	<0.005	<0.02				<0.005			2.9					
W6	5/19/2010	<0.005	<0.02				<0.005			3.4					
W6	5/21/2010	<0.005	<0.02				<0.005			1.6					
W6	5/23/2010	<0.005	<0.02				<0.005			5.2					
W6	5/25/2010	<0.005	<0.02				<0.005			5.7					
W6	5/29/2010	<0.005	<0.02				<0.005			1.4					
W6	6/2/2010	<0.005	0.97				<0.005			3.7					
W6	6/4/2010	<0.005	<0.02				<0.005			6.7					
W6	6/6/2010	<0.005	<0.02				<0.005								
W6	6/14/2010	<0.005	<0.02				<0.005			7.2					
W6	6/16/2010	<0.005	0.03				<0.005			7.9					
W6	6/18/2010	<0.005	<0.02				<0.005			7.7					
W6	6/20/2010	<0.005	0.02				<0.005			7.8					
W6	6/22/2010	<0.005	<0.02				<0.005		8.6	7.8					
W6	6/24/2010	<0.005	<0.02				<0.005		8.2	7.9					
W6	6/26/2010	<0.005	<0.02				<0.005		7.8	7.7					
W6	6/26/2010														
W6	6/28/2010	<0.005	<0.02				<0.005			7.8					
W6	7/5/2010	<0.005	<0.02				<0.005			15					
W6	8/5/2010	<0.005	<0.02				<0.005			13					
W6	9/9/2010	<0.005	<0.02				<0.005			12.9					
W6	10/8/2010	<0.005	<0.02				<0.005			13.2					
W6	8/19/2011	<0.005	1.05							8.4					
W6	7/17/2012	<0.0050	0.026		1.01					7.98					
W6	7/30/2012	0.0072	0.052		0.599					8.08					
W6	8/9/2012	0.0066	0.022												
W6	8/25/2012	<0.0050	0.045							7.68					
W6	9/13/2012	<0.0050	0.034						13.1	14.4					
W6	10/13/2012	<0.0050	0.042							9.91					
W6	11/12/2012	<0.0050	<0.020							8.74					
W6	5/14/2013	<0.0050	<0.020							22.2					
W6	6/22/2013	0.0132	0.028							7.79					
W6	7/12/2013	<0.0050	<0.020							8.29					
W6	8/22/2013	<0.0050	<0.020							7.4					
W6	9/6/2013	<0.0050	0.03							6.36					
W6	10/7/2013	<0.0050	0.042							8.17					
W6	11/16/2013	<0.0050	<0.020							8.79					
W6	12/7/2013	<0.0050	<0.020							8.77					
W7	5/27/2005	0.0018	0.0824												
W7	6/30/2005	0.0017	0.172												
W7	7/29/2005	<0.0010	0.169												
W7	8/30/2005	<0.0010	0.0226												
W7	9/28/2005	<0.0010	0.0586												
W7	10/15/2005	<0.0010	0.102												
W7	6/2/2006	0.0015	0.0449												
W7	6/8/2006														

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	N-NO2 (mg/L)	N-NO3 (mg/L)	N-TKN (mg/L)	P-PO4-T (mg/L)	P-TDP (mg/L)	P-TRP (Ortho) (mg/L)	C-TIC (mg/L)	C-TOC (mg/L)	C-DOC (mg/L)	Cation Sum (meq/L)	Anion Sum (meq/L)	Hardness (CaCO3) (mg/L)	Acidity (pH 4.5) (mg/L)	Acidity pH 8.3 (mg/L)
W7	6/15/2006														
W7	6/15/2006		0.18												
W7	6/23/2006		0.23												
W7	6/28/2006		0.112												
W7	7/7/2006		0.205												
W7	7/12/2006		0.18												
W7	7/20/2006		0.19												
W7	7/26/2006														
W7	7/26/2006		0.18												
W7	8/2/2006		0.15												
W7	8/10/2006		0.16												
W7	8/25/2006														
W7	8/25/2006		0.12												
W7	8/30/2006		0.11												
W7	9/6/2006		0.08												
W7	9/13/2006		0.094												
W7	9/20/2006		0.11												
W7	9/28/2006		0.08												
W7	10/4/2006		0.1												
W7	10/12/2006		0.12												
W7	6/5/2007	<0.05	<0.1	0.5	0.15	<0.02									
W7	6/5/2007	0.05	<0.1	0.11	0.12	<0.02									
W7	6/5/2007	<0.05	<0.1	0.4	0.13	<0.02									
W7	6/20/2007	<0.05	0.3	0.3	0.05	<0.02				8.2					
W7	7/18/2007	<0.05	0.3	0.31	<0.05	<0.02	0.05			14.3					
W7	7/18/2007	<0.05	0.3	0.36	0.05	<0.02	0.05								
W7	7/18/2007	<0.05	0.3	0.33	0.06	<0.02	0.05								
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007														
W7	8/24/2007	<0.05	0.2	0.3	0.04	<0.02									
W7	8/24/2007	<0.05	0.2	0.26	0.04	<0.02									
W7	8/24/2007	<0.05	0.2	0.24	0.04	<0.02									
W7	8/24/2007	<0.05	0.2	0.26	0.04	<0.02									
W7	8/24/2007	<0.05	0.2	0.24	0.04	<0.02									
W7	9/18/2007	<0.02	0.05	0.38	0.07	<0.02				11					
W7	10/30/2007	0.14	<0.02	0.51	<0.02	<0.02	<0.01								
W7	10/30/2007	0.14	<0.02	0.46	0.02	<0.02	<0.01								
W7	10/30/2007	0.13	<0.02	0.54	0.02	<0.02	<0.01								
W7	10/30/2007	0.13	<0.02		0.02	<0.02	<0.01								
W7	4/22/2008	0.07	<0.02	2.02	<0.05		0.09								
W7	4/22/2008	0.04	<0.02	0.34	<0.05		0.09								
W7	4/22/2008	0.07	<0.02	0.24	<0.05		0.09								
W7	4/22/2008	0.04	<0.02												
W7	6/3/2008	0.04	0.03	0.38	<0.05		0.04								
W7	6/3/2008	0.04	0.04	0.45	<0.05		0.04								
W7	6/3/2008	0.04	0.04	0.4	<0.05		0.04								
W7	6/3/2008					<0.01									
W7	6/17/2008	0.05	0.14	0.34	0.06				6.8						
W7	8/6/2008	<0.01	0.14	0.38	<0.05		0.06								
W7	8/6/2008	<0.01	0.14	0.32	0.08		0.06								
W7	8/6/2008	<0.01	0.15		<0.05										
W7	8/6/2008			0.32	0.05		0.06		9.3	9					
W7	9/18/2008	<0.01	0.03	0.51	<0.05		0.09		17.6	17.4					
W7	10/28/2008			0.33	<0.05		0.08								
W7	10/28/2008			0.37	<0.05		0.08								
W7	10/28/2008			0.32	<0.05		0.08								
W7	10/28/2008								10.7	11					
W7	4/30/2009			0.76	0.06				17.6	17.6					
W7	6/8/2009			0.49	0.03				9.4	9.2					
W7	7/28/2009														
W7	8/6/2009		0.18	0.24	0.02		<0.05								
W7	8/24/2009														
W7	8/25/2009								9.4	8.2					
W7	8/31/2009														
W7	10/7/2009														
W7	10/7/2009		0.12	0.35	0.07				8.1	6.9					
W7	10/25/2009	<0.005	0.15				<0.005		7.5	8.7					
W7	10/25/2009														
W7	4/19/2010			0.82	0.06										
W7	4/27/2010	<0.005	<0.02				0.008			22.7					
W7	4/29/2010	<0.01	<0.04				<0.005			19.3					
W7	5/1/2010	0.008	0.07				<0.005			19					
W7	5/3/2010	<0.005	<0.02				<0.005			22.2					
W7	5/5/2010	<0.005	<0.02				<0.005			14.5					
W7	5/7/2010	0.007	0.1				<0.005			12.8					
W7	5/9/2010	<0.005	0.07				<0.005			13.2					
W7	5/10/2010	0.006	0.13				<0.005			11.7					
W7	5/12/2010														
W7	5/12/2010	<0.005	0.03				<0.005			10.4					
W7	5/14/2010	<0.005	<0.02				<0.005			9.2					
W7	5/16/2010	<0.005	0.03				<0.005			8.8					
W7	5/17/2010	<0.005	<0.02				<0.005			7.2					
W7	5/19/2010	<0.005	0.06				<0.005			7.5					
W7	5/21/2010	<0.005	0.05				<0.005			8.7					
W7	5/23/2010	0.006	0.06				<0.005			8.2					
W7	5/25/2010	0.025	0.08				<0.005			8.1					
W7	5/27/2010	<0.005	0.13				<0.005			8.1					
W7	5/29/2010	0.011	0.22				<0.005			6.3					
W7	5/31/2010	0.023	0.12				<0.005			7.2					
W7	6/2/2010	0.023	0.14				<0.005			4.9					
W7	6/4/2010	0.014	0.16				<0.005			6.4					
W7	6/6/2010	0.013	0.16				<0.005								
W7	6/14/2010	<0.005	0.16				<0.005			5.9					
W7	6/16/2010	<0.005	0.17				<0.005			6.1					
W7	6/18/2010	<0.005	0.15				<0.005			6.2					
W7	6/20/2010	0.008	0.13				<0.005			7.6					
W7	6/22/2010	<0.005	0.12				<0.005		10.5	10.4					
W7	6/24/2010	0.007	0.07				<0.005		10.5	10.2					
W7	6/26/2010	<0.005	0.1				<0.005		9.2	9.3					
W7	6/28/2010	<0.005	0.13				<0.005			8.7					
W7	7/5/2010	<0.005	0.07				<0.005			16.1					
W7	8/5/2010									13.2					
W7	9/12/2010														
W7	9/12/2010	<0.005	0.03				<0.005		17.4	16.7					
W7	10/8/2010	<0.005	0.07				<0.005			12.7					
W7	4/27/2011	<0.005	<0.02						22	23					
W7	5/2/2011	<0.005	<0.02							22					
W7	6/1/2011									10.7					
W7	6/16/2011	<0.005	0.1							12.6					
W7	7/1/2011	0.007	0.05							21.1					

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	N-NO2 (mg/L)	N-NO3 (mg/L)	N-TKN (mg/L)	P-PO4-T (mg/L)	P-TDP (mg/L)	P-TRP (Ortho) (mg/L)	C-TIC (mg/L)	C-TOC (mg/L)	C-DOC (mg/L)	Cation Sum (meq/L)	Anion Sum (meq/L)	Hardness (CaCO3) (mg/L)	Acidity (pH 4.5) (mg/L)	Acidity pH 8.3 (mg/L)
W7	8/2/2011	<0.005	0.06							24.2					
W7	8/16/2011	<0.005	0.06							18.3					
W7	8/19/2011	<0.005	0.04							18.8					
W7	9/12/2011	<0.005	0.05						14	13.2					
W7	10/6/2011	<0.005	0.11						11.5	11.7					
W7	11/22/2011	<0.005	0.136						7.69	7.73					
W7	1/24/2012	<0.005	0.324							5.33					
W7	2/25/2012	<0.005	0.289				<0.0050			6.08					
W7	3/5/2012	<0.005	0.195							5.17					
W7	4/24/2012	0.0101	<0.020		0.0566				13.3	13.6					
W7	5/6/2012	<0.0050	<0.020		0.0297					17.4					
W7	5/6/2012	<0.0050	<0.020		0.0303					17					
W7	6/19/2012	0.0148	0.083		0.176				19.2	16.2					
W7	7/17/2012	<0.0050	0.089		0.078					13.3					
W7	7/30/2012	<0.0050	0.124		0.0432					12.7					
W7	8/8/2012	<0.0050	0.13							12.6					
W7	8/23/2012	<0.050	0.21		0.0322				11.3	10.3					
W7	9/13/2012	<0.0050	0.078						16.3	16.3					
W7	10/16/2012	<0.0050	0.125						12.1	12.6					
W7	11/2/2012	<0.0050	0.219		0.0227					8.19					
W7	12/29/2012	<0.0050	0.063							10.2					
W7	4/30/2013	<0.0050	0.132						4.71	4.64					
W7	5/14/2013	<0.0050	<0.020							15.3					
W7	6/19/2013	<0.0050	0.143							11.3					
W7	6/19/2013	<0.0050	0.141							10.5					
W7	7/13/2013	<0.0050	0.17							9.88					
W7	8/21/2013	<0.0050	0.157							9.84					
W7	9/6/2013	<0.0050	0.128							11					
W7	10/21/2013	<0.0050	0.168							11.9					
W7	11/16/2013	<0.0050	0.257							7.47					
W7	12/7/2013	<0.0050	0.206							7.3					
C4	5/6/2012	<0.0050	0.029		1.21					10.5					
C4	8/1/2012	0.0177	0.074		0.386					16.2					
C4	8/8/2012	0.012	0.095												
C4	8/25/2012	<0.0050	<0.020							<0.50					
C4	6/22/2013	0.0066	0.371							17.4					
C4	7/19/2013	0.0069	0.082												
C4	8/22/2013	0.0084	0.086							18.9					
C4	9/6/2013	<0.0050	0.07							15.1					
C4	10/18/2013	<0.0050	0.069							12.7					
C4	10/18/2013	<0.0050	0.065							12.5					
C4	11/17/2013	<0.050	<0.20							31.6					
C10	5/12/2012	0.0114	0.047		2.25					13.3					
C10	7/30/2012	0.0229	0.184		0.903					17.8					
C10	7/30/2012														
C10	8/10/2012	0.0111	0.184												
C10	8/27/2012	0.0086	0.095							17.8					
C10	6/22/2013	0.0198	0.152							15.3					
C10	7/23/2013	<0.050	<0.20												
C10	8/22/2013	0.0072	0.135							17.2					
C10	9/8/2013	0.0076	0.175							16					
C10	9/8/2013	0.007	0.192							16.2					
C10	10/18/2013	<0.0050	0.167							14.3					

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

Station Name	Sample Date	Lab pH (pH units)	Lab Conductance (µS/cm)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	Hardness-E (mg/L)	Turbidity (NTU)	Temp-obs-pH (°C)	IB (%)	TDS (mg/L)	TSS (mg/L)	Total Alkalinity (mg/L)	Alkalinity-P (mg/L)	Alkalinity-OH (mg/L)	Alkalinity-Carb (mg/L)	Alkalinity-Bicrb (mg/L)
W1	5/27/2005	8.05	195		97.3					137	26.5	78.7			89.2	
W1	6/30/2005	8.17	290							189	<3.0	141			141	
W1	7/28/2005	8.03	312		157					189	<3.0				151	
W1	8/29/2005	8.28	266		138					186	4	138			138	
W1	9/28/2005	7.59	244		134					178	9.5	117			117	
W1	10/15/2005	8.13	253		140					181	14.5				121	
W2	5/27/2005	8.06	198		99.8					138	20.5				89.3	
W2	5/27/2005	8.03	148		72.2					80	79	67.1				
W2	6/30/2005	8.17	287							190	<3.0	140			140	
W2	7/28/2005	8.01	305		153					194	<3.0				146	
W2	8/30/2005	8.26	268		138					193	3.5	143			143	
W2	9/28/2005	7.9	243		134					170	9	115			115	
W2	10/15/2005	8.11	249		137					179	5.5				120	
W3	5/27/2005	7.99	188		93.9					141	8.5	61.2			78.7	
W3	6/30/2005	8.22	341							224	<3.0	153			153	
W3	7/29/2005	8.06	322		166					198	25.7				143	
W3	8/29/2005	8.24	267		135					182	7.5	122			122	
W3	9/28/2005	7.99	238		129					174	8	101			101	
W3	10/15/2005	8.08	245		134					176	<3.0				106	
W8	5/27/2005	7.78	113		62.3					111	<3.0	174			49.1	
W8	6/30/2005	7.83	176							144	<3.0	89			89	
W8	7/29/2005	7.45	203		116					141	24.7				104	
W8	8/30/2005	8.12	179		99.8					147	<3.0	99.2			99.2	
W8	9/28/2005	7.96	165		95					135	<3.0	74.7			74.7	
W8	10/15/2005	7.06	190		99.4					140	5.5				87.1	
W9	5/27/2005	7.56	93.2		49.4					104	<3.0	142			32	
W9	6/30/2005	7.63	145							126	<3.0	56.1			56.1	
W9	7/29/2005	7.51	176		95.3					129	3.7				74.2	
W9	8/29/2005	7.84	146		73.6					126	<3.0	57.7			57.7	
W9	9/28/2005	7.62	125		67.8					118	<3.0	44			44	
W9	10/15/2005	7.19	121		64.4					111	<3.0				51.8	
W10	5/27/2005	7.47	75.4		46.3					115	<3.0				36.7	
W10	5/27/2005															
W10	6/30/2005	7.31	84.2							116	6	41.5			41.5	
W10	7/29/2005	7.34	111		54.9					105	4.2				58.2	
W10	8/29/2005	7.74	145		70					117	13	79.1			79.1	
W10	9/28/2005	7.77	181		101					141	14.5	92.3			92.3	
W10	10/15/2005	7.3	175		100					138	6.5				96.4	
W6	5/27/2005	7.91	127		65.4					103	13.5	32			61.2	
W6	6/30/2005	7.41	174							123	11.5	99.1			99.1	
W6	7/29/2005	8.01	319		171					198	22.2				147	
W6	8/30/2005	8.17	200		106					147	<3.0	109			109	
W6	9/28/2005	7.95	192		106					133	3.5	93			93	
W6	10/15/2005	7.92	180		98.3					128	<3.0				99.3	
W6	6/2/2006	7.64	112							96	115	76.6				
W6	8/25/2006	8.01	223	120						160		120		<5	<6	146
W6	9/6/2006	7.86	212	110							4	112		<5	<6	137
W6	6/20/2007	7.77	151		81		0.6			130	16	83		<5	<6	100
W6	9/18/2007	7.95	194		93		0.2			172	8	112		<5	<6	137
W6	6/17/2008	7.82	175		86		0.9			120	17	93		<5	<6	100
W6	9/18/2008	7.8	174				<0.1			150	4	87		<5	<6	100
W6	4/14/2009	7.92	624	248			24			464	138	201		<5	<6	240
W6	4/30/2009	7.34	104	56			39			112	190	41		<5	<6	50
W6	7/28/2009	7.86	338	148			27			228	38	92		<5	<6	100
W6	8/6/2009															
W6	8/6/2009	8.05	413	177			0.3			296	<2	155		<5	<6	190
W6	8/18/2009			173												
W6	8/25/2009															
W6	10/7/2009	7.86	478	195			0.5			314	<2	95		<5	<6	100
W6	10/7/2009	7.83	475	195			0.6			352	<3	96		<5	<6	100
W6	10/25/2009	8.1	501	207	219		1.8			270	<4	120	<0.5	<0.5	<0.5	150
W6	4/19/2010	7.93	292		135		38			196	29	117		<5	<6	140
W6	5/16/2010	8.1	200	98	97.2		0.7			130	<4	100	<0.5	<0.5	<0.5	120
W6	5/17/2010	7.8	152	73.7	68.8		1.6			120	<4	71	<0.5	<0.5	<0.5	87
W6	5/19/2010	8.1	192	95.5	101		1.2			120	3	90	<0.5	<0.5	<0.5	110
W6	5/21/2010	8	160	82.6	88.1		2.9			84	10	80	<0.5	<0.5	<0.5	97
W6	5/23/2010	8.1	219	102	96.4		2			150	9	100	<0.5	<0.5	<0.5	120
W6	5/25/2010	7.9	221	108	120		1.6			120	4	100	<0.5	<0.5	<0.5	120
W6	5/29/2010	8	141	69.2	70.1		2.8			84	10	74	<0.5	<0.5	<0.5	90
W6	6/2/2010	8.4	466	221	243		0.7			280	2	200	5.2	<0.5	6.3	230
W6	6/4/2010	8.1	189	81	92.6		1.7			96	1	100	<0.5	<0.5	<0.5	120
W6	6/6/2010	8	187	98.3	98.8		3.1			120	5	100	<0.5	<0.5	<0.5	130
W6	6/14/2010	8.2	184	92.4	94.5		3.6			130	13	100	<0.5	<0.5	<0.5	120
W6	6/16/2010	8.2	188	92	90.8		2.7			120	5	100	<0.5	<0.5	<0.5	130
W6	6/18/2010	8.2	191	93.9	89.2		1.7			120	6	100	<0.5	<0.5	<0.5	130
W6	6/20/2010	8.2	182	88.8	92.7		2.7			110	5	100	<0.5	<0.5	<0.5	120
W6	6/22/2010	8.17	190	92.6	94.3		3.4			120	8	100	<0.5	<0.5	<0.5	120
W6	6/24/2010	8.11	189	92.6	74.4		2.1			96	4	100	<0.5	<0.5	<0.5	120
W6	6/26/2010	8.11	189	90.3	90.7		3			92	5	100	<0.5	<0.5	<0.5	120
W6	6/26/2010															
W6	6/28/2010	7.87	189	89.5	91.4		6.8			130	19	99	<0.5	<0.5	<0.5	120
W6	7/5/2010	7.95	170	79.5	86.9		23			130	60	87	<0.5	<0.5	<0.5	110
W6	8/5/2010	8.06	207	108	101		2.2			130	8	110	<0.5	<0.5	<0.5	130
W6	9/9/2010	8.07	210	108	107		10.2			160	31	110	<0.5	<0.5	<0.5	130
W6	10/8/2010	8.07	209	103	100		6.7			140	46	110	<0.5	<0.5	<0.5	130
W6	8/19/2011	8.3	485	225	214					280	2	190	<0.5	<0.5	<0.5	240
W6	7/17/2012	7.98	201	104	163					122	1130	103	<0.50	<0.50	<0.50	126
W6	7/30/2012	8.09	202	101	145					124	614	103	<0.50	<0.50	<0.50	126
W6	8/9/2012	8.17	220	112	136					144	566	116	<0.50	<0.50	<0.50	141
W6	8/25/2012	8	212	103	161					138	1160	109	<0.50	<0.50	<0.50	133
W6	9/13/2012	8.04	174	87.7	93.1					118	50.3	87.1	<0.50	<0.50	<0.50	106
W6	10/13/2012	7.92	193		117					150	44.7	99.1	<0.50	<0.50	<0.50	121
W6	11/12/2012	7.93	275	153	155					162	2.2	138	<0.50	<0.50	<0.50	168
W6	5/14/2013	7.48	55.1	31.4	33.8					84	13.1	25.5	<0.50	<0.50	<0.50	31.1
W6	6/22/2013	8.07	195	94.9	115					118	172	97.8	<0.50	<0.50	<0.50	119
W6	7/12/2013	7.9	180	88.2	95.3					136	25.2	93.8	<0.50	<0.50	<0.50	115
W6	8/22/2013	8.04	205	102	106					144	19.7	107	<0.50	<0.50	<0.50	130
W6	9/6/2013	7.95	204	101	114					130	5.2	105	<0.50	<0.50	<0.50	129
W6	10/7/2013	8.08	191	107	106					176	74.4	98.1	<0.50	<0.50	<0.50	120
W6	11/16/2013	8.11	245	119	123					120	33.4	121	<0.50	<0.50	<0.50	147
W6	12/7/2013	8.21	326	159	182					194	3.6	164	<0.50	<0.50	<0.50	200
W7	5/27/2005	7.95	152		76.6					116	30.5	36.7			68.4	
W7	6/30/2005	7.87	233							153	14	117			117	
W7	7/29/2005	7.18	323		167					197	19.7				146	
W7	8/30/2005	8.19	211		109					148	3	113			113	
W7	9/28/2005	8.02	190		107					142	361	95.9			95.9	
W7	10/15/2005	6.75	194		106					145	7.5				107	
W7	6/2/2006	7.61	98.8							104	705	49.4				
W7	6/8/2006															

Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

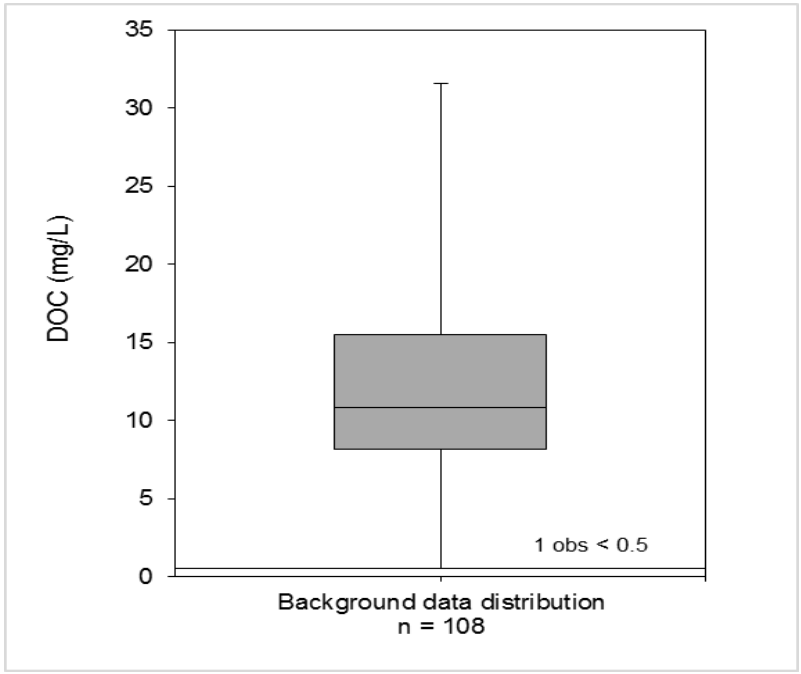
Station Name	Sample Date	Lab pH (pH units)	Lab Conductance (µS/cm)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	Hardness-E (mg/L)	Turbidity (NTU)	Temp-obs-pH (°C)	IB (%)	TDS (mg/L)	TSS (mg/L)	Total Alkalinity (mg/L)	Alkalinity-P (mg/L)	Alkalinity-OH (mg/L)	Alkalinity-Carb (mg/L)	Alkalinity-Bicrb (mg/L)
W7	6/15/2006															
W7	6/15/2006	7.86	193	100						114	53	95		<5	<6	116
W7	6/23/2006	7.78	246	122						176	328	130		<5	<6	159
W7	6/28/2006	7.9	213	107						120	103	106		<5	<6	129
W7	7/7/2006	7.91	205	110						184	3	120		<5	<6	146
W7	7/12/2006	7.77	238	110							<2	120		<5	<6	146
W7	7/20/2006	7.88	233	130							2	122		<5	<6	149
W7	7/26/2006															
W7	7/26/2006	7.87	259	128							9	121		<5	<6	147
W7	8/2/2006	7.79	256	128							4	125		<5	<6	152
W7	8/10/2006	7.8	259	126							<2	125		<5	<6	152
W7	8/25/2006															
W7	8/25/2006	7.92	261	129						180		131		<5	<6	160
W7	8/30/2006	7.79	258	132						200	<2	129		<5	<6	157
W7	9/6/2006	7.88	258	130							4	129		<5	<6	157
W7	9/13/2006	7.84	254	130							5	128		<5	<6	157
W7	9/20/2006	7.85	263	132							3	127		<5	<6	154
W7	9/28/2006	7.26	266	129						182		128		<5	<6	156
W7	10/4/2006	7.94	258	126							2	122		<5	<6	149
W7	10/12/2006	7.39	266	132						164	4	130		<5	<6	158
W7	6/5/2007	7.98	208		108		2.4			162	76	118		<5	<6	144
W7	6/5/2007	7.97	207		120		3.1			158	76	121		<5	<6	147
W7	6/5/2007	7.97	207		111		2.2			166	73	105		<5	<6	128
W7	6/20/2007	7.84	230	110			0.5			164	6	116		<5	<6	142
W7	7/18/2007	7.72	250	128	126		0.6			170	4	128		<5	<6	156
W7	7/18/2007	7.8	249		125		1			182	7	128		<5	<6	156
W7	7/18/2007	7.83	249		125		1.4			184	6	128		<5	<6	157
W7	8/24/2007															
W7	8/24/2007															
W7	8/24/2007															
W7	8/24/2007	8.04	246		122		<0.1			198	4	139		<5	<6	170
W7	8/24/2007	8.05	245		123		<0.1			192	4	136		<5	<6	166
W7	8/24/2007	8.04	244		128		<0.1			186	6	136		<5	<6	165
W7	9/18/2007	8	224	110			0.2			182	20	124		<5	<6	151
W7	10/30/2007	7.74	268		130		0.9			176	3	128		<5	<6	157
W7	10/30/2007	7.74	274		128		1.9			182	10	131		<5	<6	160
W7	10/30/2007	7.74	269		130		1.9			166	7	128		<5	<6	157
W7	10/30/2007	7.68	272				1.2			172	6	128		<5	<6	157
W7	4/22/2008	8.4	364	186			0.3			264	<2	187		<5	<6	218
W7	4/22/2008	8.41	362	193			0.3			274	<2	185		<5	<6	215
W7	4/22/2008	8.4	366	182			0.4			270	<2	186		<5	<6	216
W7	4/22/2008	8.4	367				0.5			262	<2	188		<5	<6	218
W7	6/3/2008	8.01	201		97		1.3			134	<2	94		<5	<6	100
W7	6/3/2008	8.01	203		96		1.3			132	19	95		<5	<6	100
W7	6/3/2008	8.03	205		101		1.1			146	26	95		<5	<6	100
W7	6/3/2008			92												
W7	6/17/2008	7.95	237				1			156	10	115		<5	<6	140
W7	8/6/2008	7.9	234		123		1.4			162	92	128		<5	<6	160
W7	8/6/2008	7.91	235		123		1.4			168	139	127		<5	<6	150
W7	8/6/2008	7.96	233		125		1.2			170	85	127		<5	<6	160
W7	8/6/2008			123												
W7	9/18/2008	7.89	170				1.8			170	28	82		<5	<6	100
W7	10/28/2008	7.84	217				0.6			192	4	108		<5	<6	130
W7	10/28/2008	7.83	217				0.7			188	6	108		<5	<6	130
W7	10/28/2008	7.85	218				0.8			168	7	108		<5	<6	130
W7	10/28/2008			123												
W7	4/30/2009	6.84	36	20			1.6			52	<2	12		<5	<6	20
W7	6/8/2009	7.9	208	93			1.4			156	32	95		<5	<6	100
W7	7/28/2009	8.07	270	137			0.2			184	4	128		<5	<6	160
W7	8/6/2009	8.35	260	121			0.2			202	2	159		<5	<6	190
W7	8/24/2009															
W7	8/25/2009															
W7	8/31/2009			110												
W7	10/7/2009															
W7	10/7/2009	7.87	249	123			0.2			180	7	116		<5	<6	140
W7	10/25/2009	8.1	266	126	141		0.7			130	<4	130	<0.5	<0.5	<0.5	160
W7	10/25/2009	8	264				0.3			110	<4					
W7	4/19/2010	7.81	167		80		1.2			142	<7	82		<5	<6	100
W7	4/27/2010	7.6	64	36.3	35		1.2			62	<4	32	<0.5	<0.5	<0.5	39
W7	4/29/2010	7.5	66	36.2	36.2		1.8			62	<4	31	<0.5	<0.5	<0.5	38
W7	5/1/2010	7.7	78	40.8	43.6		8.6			74	15	40	<0.5	<0.5	<0.5	49
W7	5/3/2010	7.7	85	44.9	46.2		7.9			80	12	42	<0.5	<0.5	<0.5	51
W7	5/5/2010	7.8	102	49.1	55.1		8.7			76	12	53	<0.5	<0.5	<0.5	64
W7	5/7/2010	7.8	113	56.9	56		4.1			62	18	58	<0.5	<0.5	<0.5	70
W7	5/9/2010	8	180	79.9	71		2.2			130	<4	86	<0.5	<0.5	<0.5	110
W7	5/10/2010	8.1	163	80.9	90.3		1			130	<4	83	<0.5	<0.5	<0.5	100
W7	5/12/2010															
W7	5/12/2010	8	166	81.6	84.6		1.2			100	<4	83	<0.5	<0.5	<0.5	100
W7	5/14/2010	8.1	149	72.6	71.4		7.7			130	15	73	<0.5	<0.5	<0.5	89
W7	5/16/2010	8.3	590	85.6	76.3		3.8			120	12	260	<0.5	<0.5	<0.5	320
W7	5/17/2010	8.2	160	80.4	75.6		4			130	11	80	<0.5	<0.5	<0.5	98
W7	5/19/2010	8.2	195	98.6	103		1.5			120	4	98	<0.5	<0.5	<0.5	120
W7	5/21/2010	8.1	191	91.5	95.6		6.5			120	22	96	<0.5	<0.5	<0.5	120
W7	5/23/2010	8.1	204	98.1	103		29.3			150	33	100	<0.5	<0.5	<0.5	130
W7	5/25/2010	7.9	211	125	124		85.5			140	140	100	<0.5	<0.5	<0.5	130
W7	5/27/2010	8	231	113	148		307			140	350	110	<0.5	<0.5	<0.5	140
W7	5/29/2010	8.2	232	111	129		170			160	200	110	<0.5	<0.5	<0.5	140
W7	5/31/2010	8.2	236	109	130		155			160	290	120	<0.5	<0.5	<0.5	140
W7	6/2/2010	8.2	243	113	135		68.2			140	81	120	<0.5	<0.5	<0.5	150
W7	6/4/2010	8.2	240	111	125		51.6			140	60	120	<0.5	<0.5	<0.5	150
W7	6/6/2010	8.2	250	128	133		45.2			170	39	130	<0.5	<0.5	<0.5	150
W7	6/14/2010	8.29	250	122	128		17			150	43	130	<0.5	<0.5	<0.5	160
W7	6/16/2010	8.3	259	120	128		6.4			150	17	130	<0.5	<0.5	<0.5	160
W7	6/18/2010	8.17	256	122	123		7.8			140	28	130	<0.5	<0.5	<0.5	160
W7	6/20/2010	8.2	240	112	118		18.9			140	33	120	<0.5	<0.5	<0.5	150
W7	6/22/2010	8.03	208	97.6	115		134			140	300	110	<0.5	<0.5	<0.5	130
W7	6/24/2010	8.03	214	99.7	116		24.5			130	64	110	<0.5	<0.5	<0.5	130
W7	6/26/2010	8.26	230	106	106		8.2			120	21	120	<0.5	<0.5	<0.5	140
W7	6/28/2010	8.03	248	114	107		3			150	9	120	<0.5	<0.5	<0.5	150
W7	7/5/2010	8.06	202	92.5	106		19			150	53	100	<0.5	<0.5	<0.5	120
W7	8/5/2010			118	116					170						
W7	9/12/2010				105											
W7	9/12/2010	7.98	212	107	102		3.4			210	11	100	<0.5	<0.5	<0.5	130
W7	10/8/2010	8.1	233	111	126		1.6			150	3	120	<0.5	<0.5	<0.5	140
W7	4/27/2011	7.41	79	42.6	42.6					72	<1	34	<0.5	<0.5	<0.5	41
W7	5/2/2011	7.11	57	31.8	33.1					40	7	27	<0.5	<0.5	<0.5	33
W7	6/1/2011	8.09	196	92.3	94					120	23	96	<0.5	<0.5	<0.5	120
W7	6/16/2011	8	195	97.4	108					130	85	92	<0.5	<0.5	<0.5	110
W7	7/1/2011	7.76	146	77.1	87.5					120	150	73	<0.5	<0.5	<0.5	89

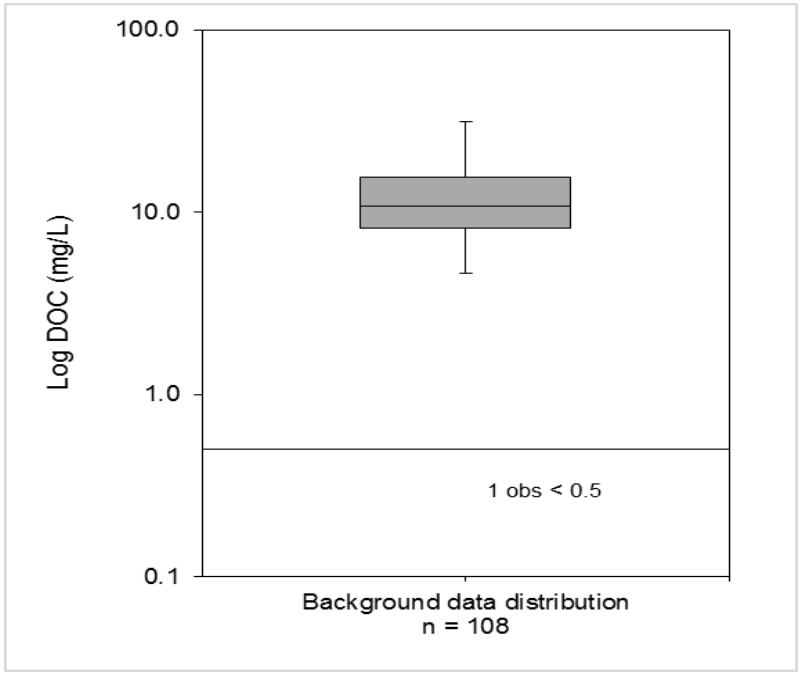
Table B.1: Raw water quality data from Minto Mine, 2005 - 2013.

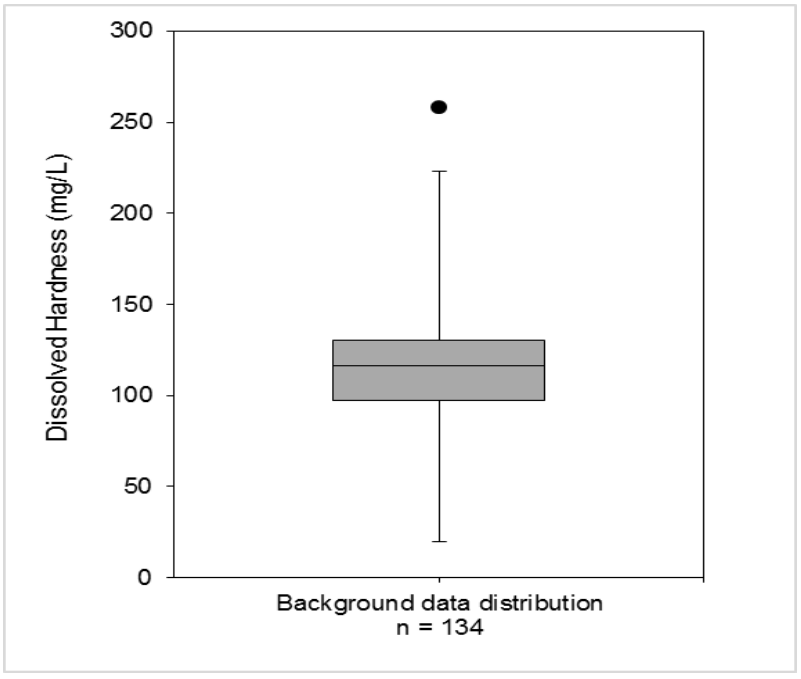
Station Name	Sample Date	Lab pH (pH units)	Lab Conductance (µS/cm)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	Hardness-E (mg/L)	Turbidity (NTU)	Temp-obs-pH (°C)	IB (%)	TDS (mg/L)	TSS (mg/L)	Total Alkalinity (mg/L)	Alkalinity-P (mg/L)	Alkalinity-OH (mg/L)	Alkalinity-Carb (mg/L)	Alkalinity-Bicrb (mg/L)
W7	8/2/2011	7.67	113	58.9	85.8					110	400	54	<0.5	<0.5	<0.5	66
W7	8/16/2011	7.9	191	97.6	96.8					150	22	94	<0.5	<0.5	<0.5	110
W7	8/19/2011	8.12	185	93	93.4					140	22	92	<0.5	<0.5	<0.5	110
W7	9/12/2011	8.03	217	118	119					170	1	110	<0.5	<0.5	<0.5	130
W7	10/6/2011	8.07	233	110	113					150	<4	120	<0.5	<0.5	<0.5	140
W7	11/22/2011	7.96	280	148	144					156	26.7	134	<0.50	<0.50	<0.50	163
W7	1/24/2012	8.13	349	169	158					226	<4.0	165	<0.50	<0.50	<0.50	201
W7	2/25/2012	7.99	339	165	160					216	<4.0	155	<0.50	<0.50	<0.50	189
W7	3/5/2012	8.17	343	161	159					176	3.2	154	<0.50	<0.50	<0.50	188
W7	4/24/2012	7.63	75	38.7	37					100	11.2	36.2	<0.50	<0.50	<0.50	44.1
W7	5/6/2012	6.05	93.2	46.1	44.8					80	2.4	27.3	<0.50	<0.50	<0.50	33.3
W7	5/6/2012	5.74	92.3	46.4	44.7					88	4.2	8.45	<0.50	<0.50	<0.50	10.3
W7	6/19/2012	8.12	178	91.8	105					128	165	87.4	<0.50	<0.50	<0.50	107
W7	7/17/2012	8.15	254	128	124					164	34.1	128	<0.50	<0.50	<0.50	156
W7	7/30/2012	8.2	272	135	128					180	9.2	138	<0.50	<0.50	<0.50	168
W7	8/8/2012	8.26	278	141	141					192	3.3	143	<0.50	<0.50	<0.50	175
W7	8/23/2012	8.25	285	142	151					200	1.5	147	<0.50	<0.50	<0.50	179
W7	9/13/2012	8.13	230	116	121					154	15.3	115	<0.50	<0.50	<0.50	140
W7	10/16/2012	8.17	261	129	143					166	9.9	132	<0.50	<0.50	<0.50	161
W7	11/2/2012	7.93	319	158	153					214	2.2	150	<0.50	<0.50	<0.50	183
W7	12/29/2012	8.13	444	223	206					262	<1.0	207	<0.50	<0.50	<0.50	253
W7	4/30/2013	8.27	387	192	198					246	5.8	187	<0.50	<0.50	<0.50	228
W7	5/14/2013	7.58	61.3	33.3	36.4					70	55.3	29.4	<0.50	<0.50	<0.50	35.9
W7	6/19/2013	8.08	238	113	114					152	11.2	113	<0.50	<0.50	<0.50	138
W7	6/19/2013	8.07	237	114	115					164	10.1	113	<0.50	<0.50	<0.50	138
W7	7/13/2013	8.1	263	132	137					156	5	127	<0.50	<0.50	<0.50	155
W7	8/21/2013	8.23	286	140	144					172	3.8	143	<0.50	<0.50	<0.50	175
W7	9/6/2013	8.02	280	138	154					280	11.8	140	<0.50	<0.50	<0.50	170
W7	10/21/2013	8.16	288	147	126					168	7.4	138	<0.50	<0.50	<0.50	168
W7	11/16/2013	8.24	324	153	164					164	1.5	153	<0.50	<0.50	<0.50	187
W7	12/7/2013	8.14	354	168	189					220	<1.0	166	<0.50	<0.50	<0.50	203
C4	5/6/2012	7.59	88.8	43.9	119					84	1260	40.1	<0.50	<0.50	<0.50	48.9
C4	8/1/2012	8.04	228	117	133					166	246	115	<0.50	<0.50	<0.50	140
C4	8/8/2012	8.14	248	125	142					170	253	126	<0.50	<0.50	<0.50	153
C4	8/25/2012	7.94	254	130	173					192	855	128	<0.50	<0.50	<0.50	156
C4	6/22/2013	7.97	253	126	222					204	886	120	<0.50	<0.50	<0.50	147
C4	7/19/2013	8.12	253	128	200					184	487	125	<0.50	<0.50	<0.50	152
C4	8/22/2013	7.98	245	123	199					182	784	118	<0.50	<0.50	<0.50	144
C4	9/6/2013	7.93	248	125	166					176	172	120	<0.50	<0.50	<0.50	146
C4	10/18/2013	8.18	260	129	142					214	16.3	126	<0.50	<0.50	<0.50	154
C4	10/18/2013	8.24	260	129	139					220	16.3	126	<0.50	<0.50	<0.50	154
C4	11/17/2013	7.9	477	258	264					272	22.4	244	<0.50	<0.50	<0.50	298
C10	5/12/2012	7.73	136	67.2	215					132	2210	65.1	<0.50	<0.50	<0.50	79.4
C10	7/30/2012	8.23	254	130	207					180	995	132	<0.50	<0.50	<0.50	161
C10	7/30/2012															
C10	8/10/2012	8.11	273	135	245					216	1320	146	<0.50	<0.50	<0.50	178
C10	8/27/2012	8.37	284	141	264					184	1650	154	2.03	<0.50	2.44	183
C10	6/22/2013	8.26	276	148	151					192	43.4	143	<0.50	<0.50	<0.50	175
C10	7/23/2013	8.19	251	125	156					174	199	133	<0.50	<0.50	<0.50	162
C10	8/22/2013	8.23	309	160	168					218	65.5	159	<0.50	<0.50	<0.50	194
C10	9/8/2013	7.94	287	150	163					186	25.4	147	<0.50	<0.50	<0.50	180
C10	9/8/2013	8.13	289	150	162					206	21.7	151	<0.50	<0.50	<0.50	184
C10	10/18/2013	8.25	263	136	160					212	61.2	136	<0.50	<0.50	<0.50	166

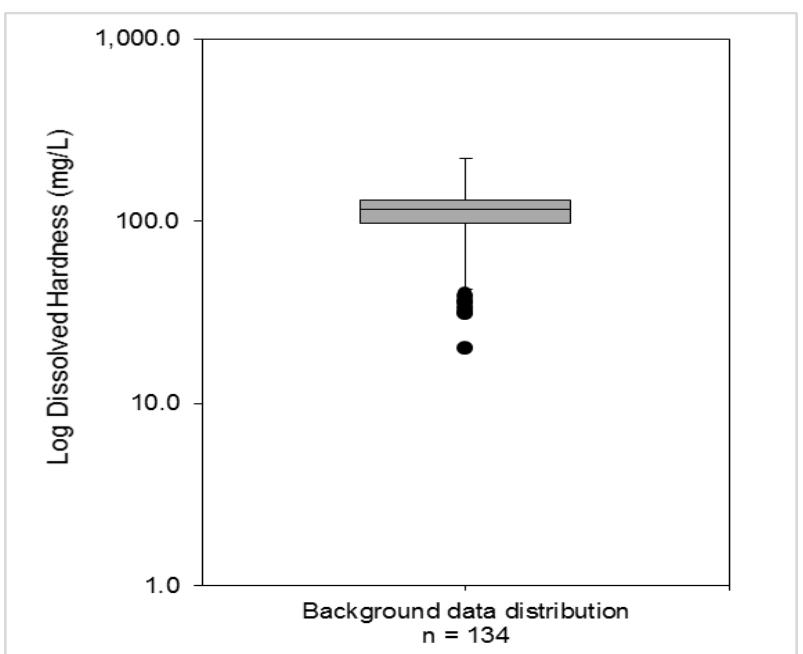
APPENDIX C

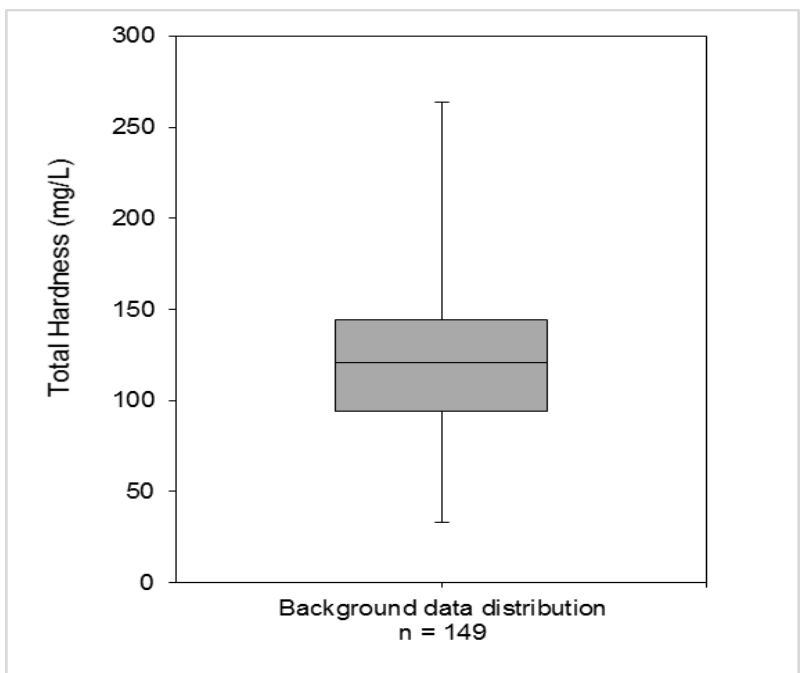
BOXPLOT OUTLIER EXAMINATION

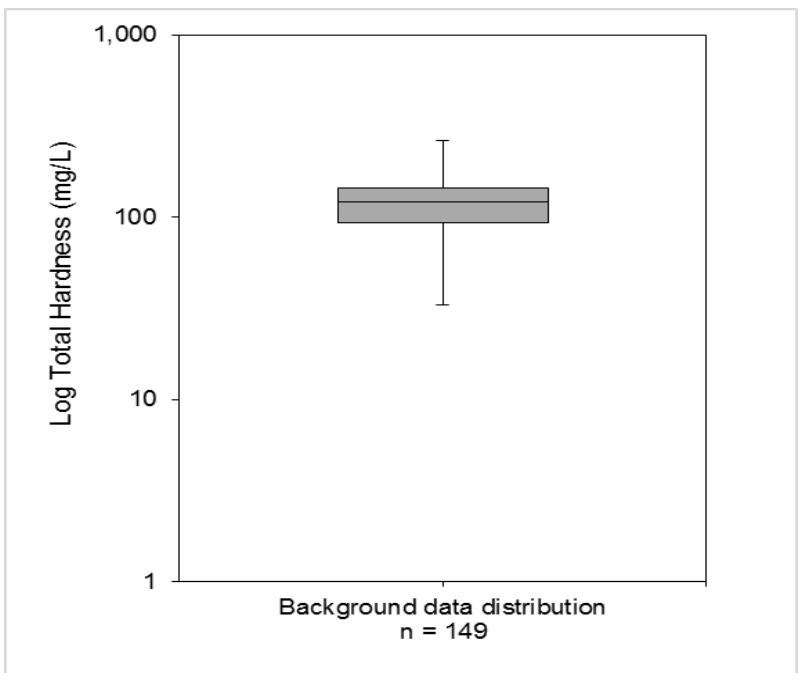


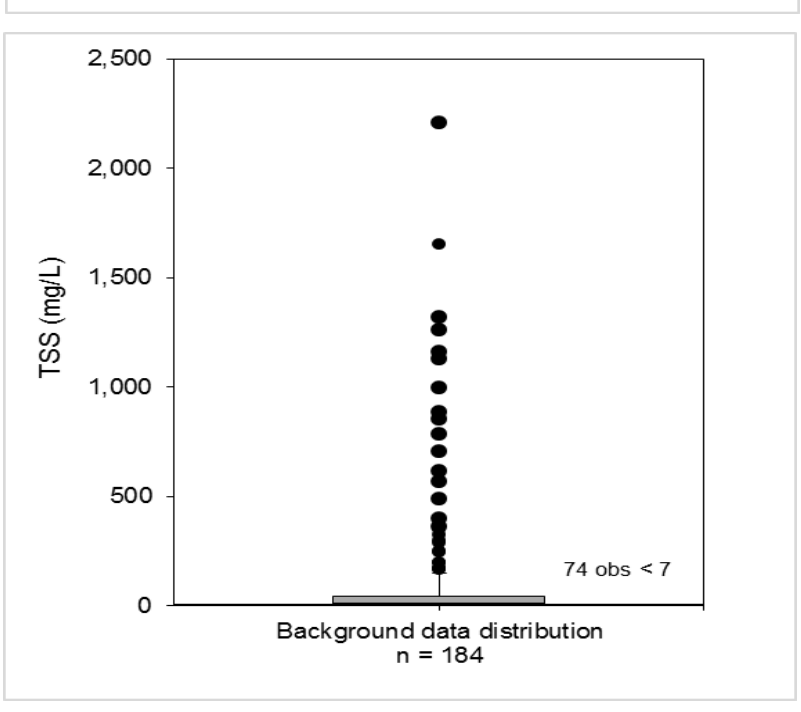












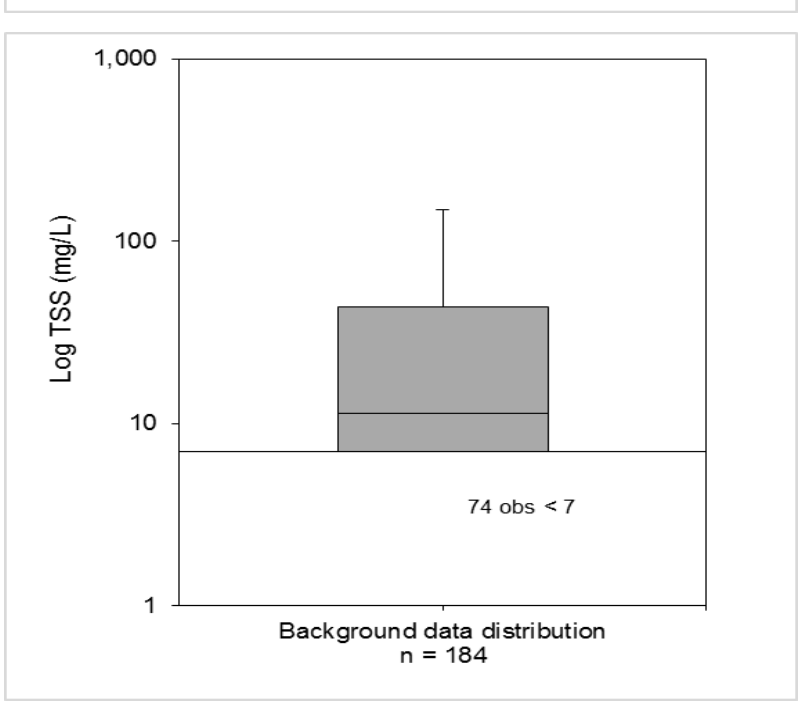
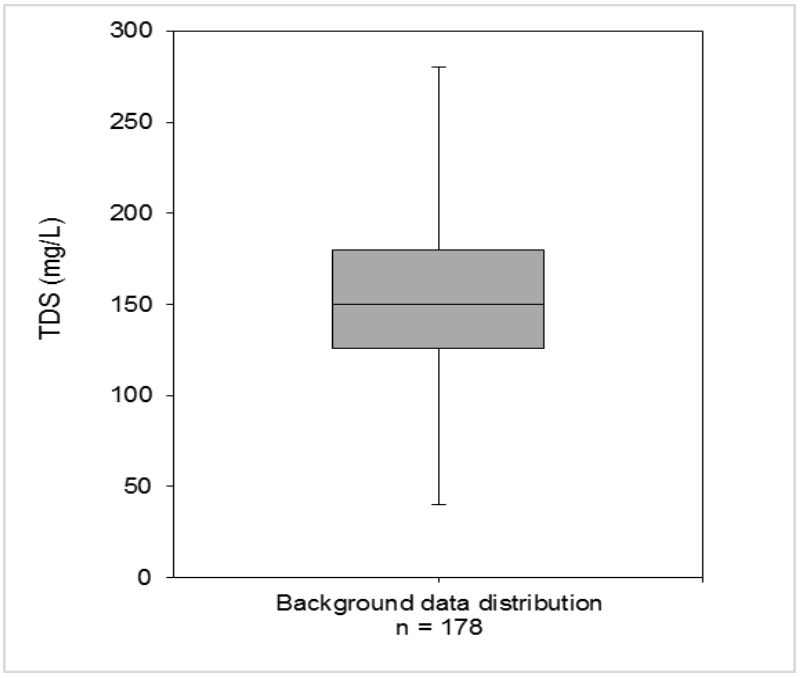
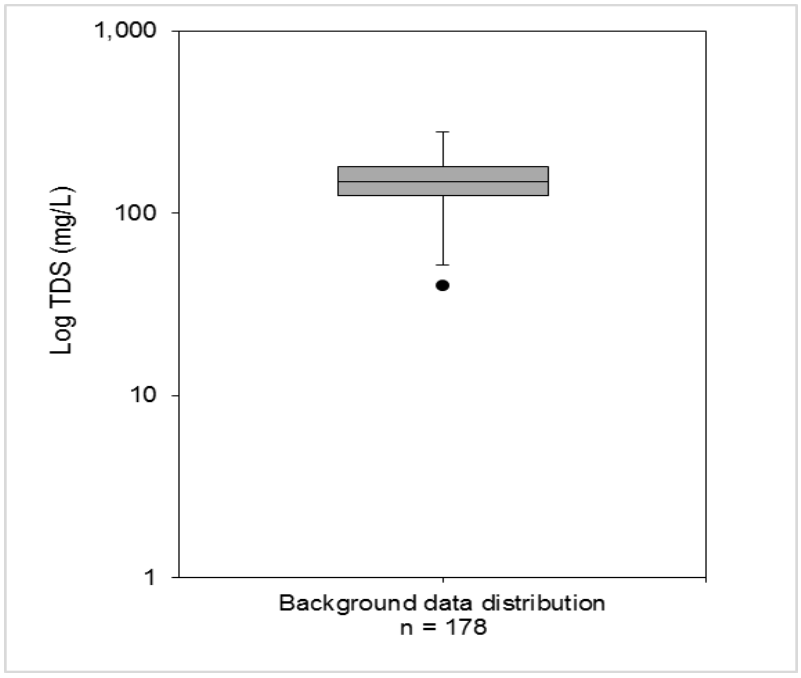
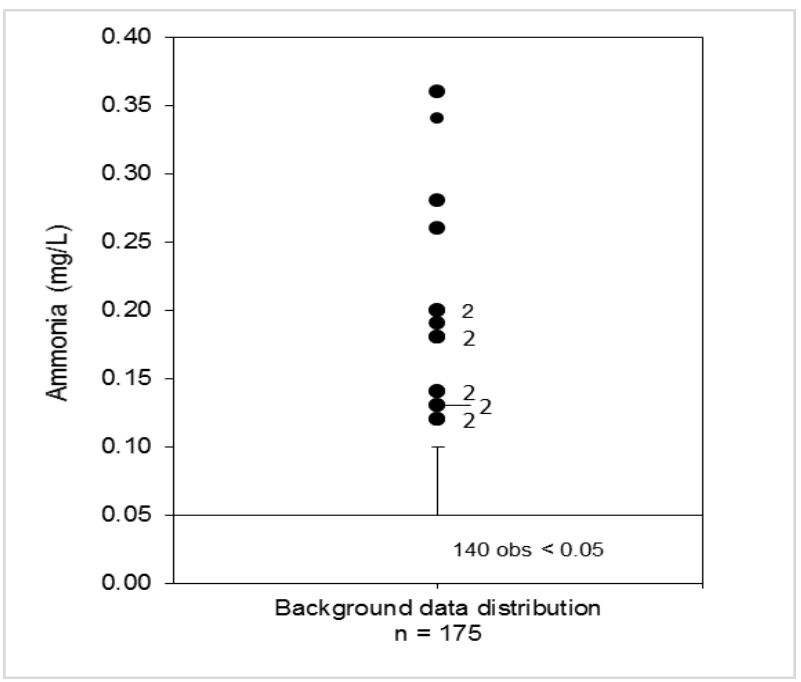


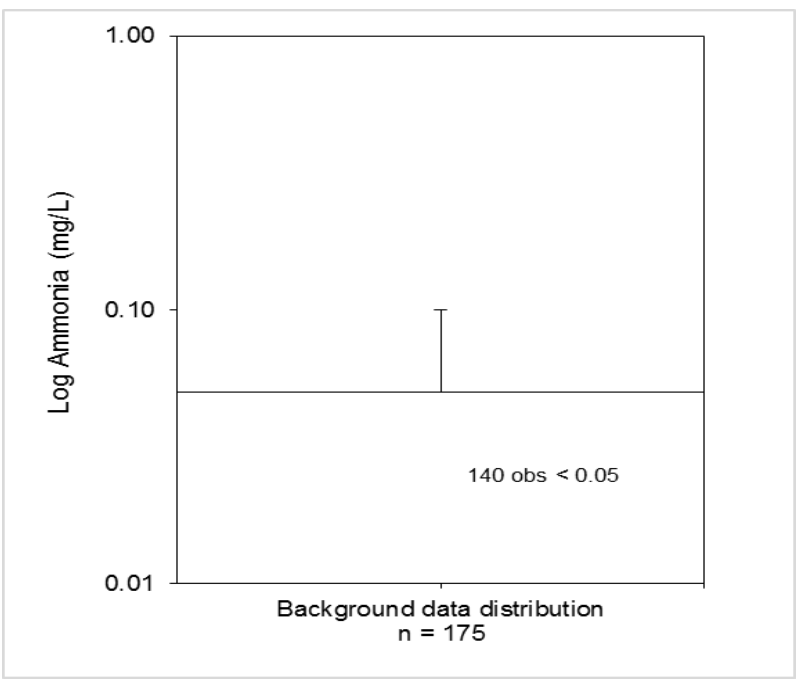
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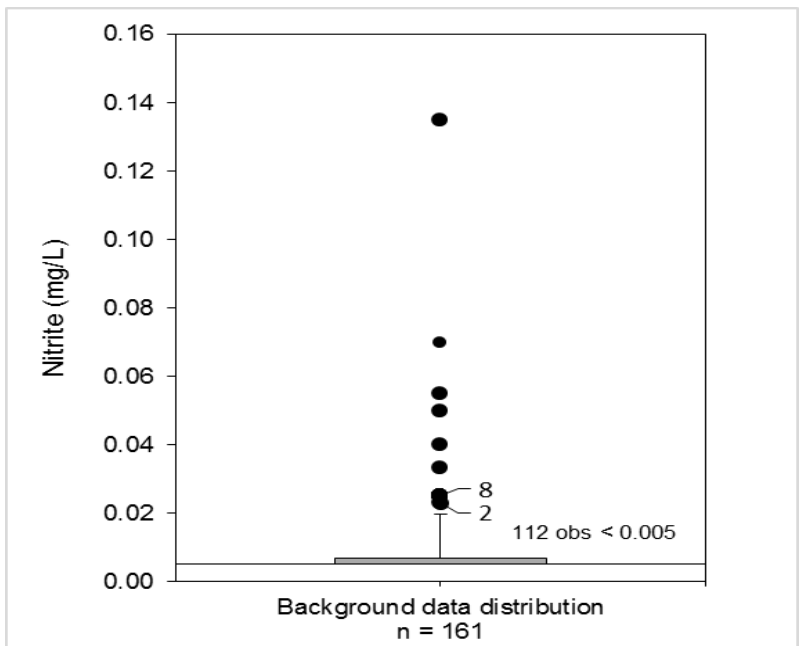
NOTE: Possible outliers were only considered if they were detected on both non-transformed and log transformed scales

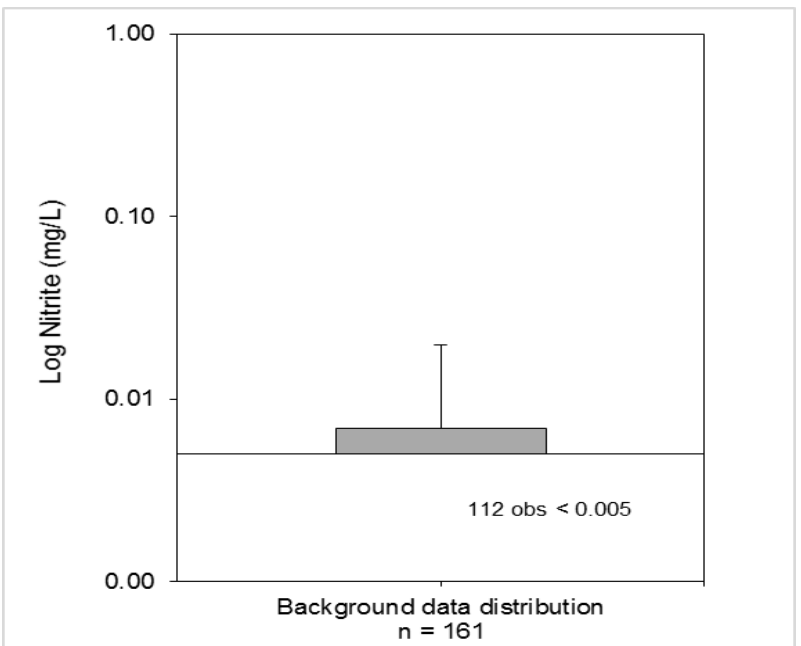


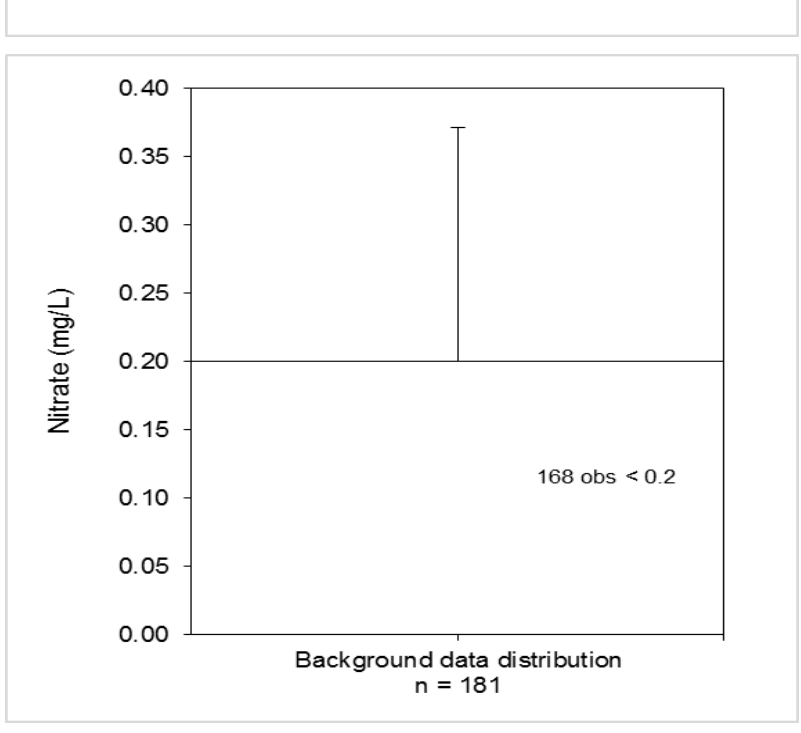












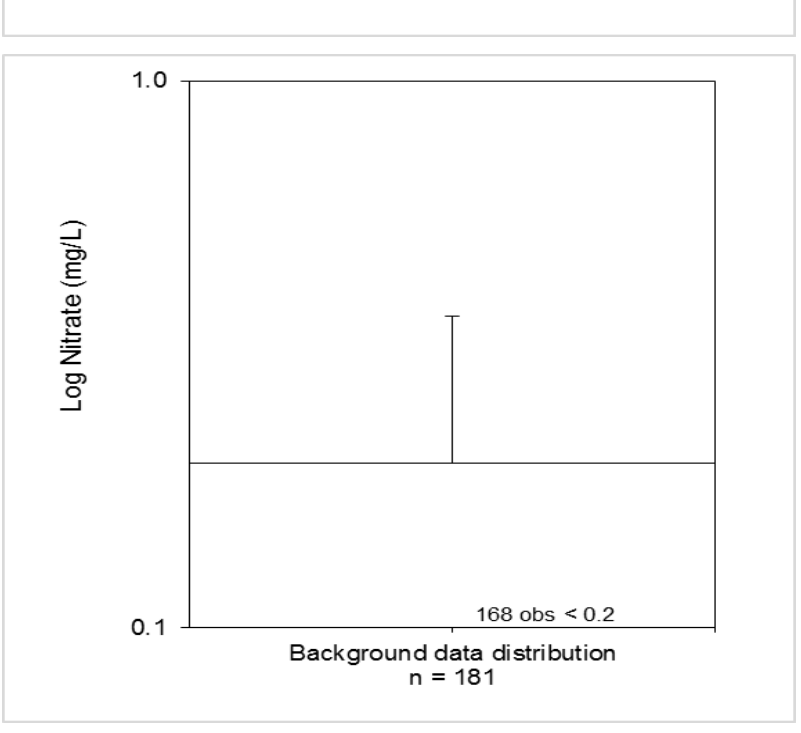
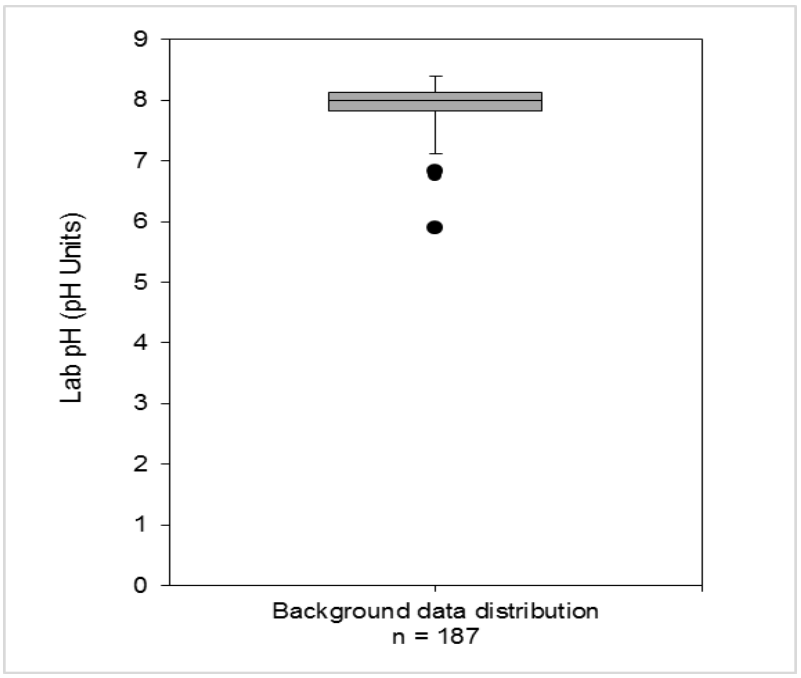
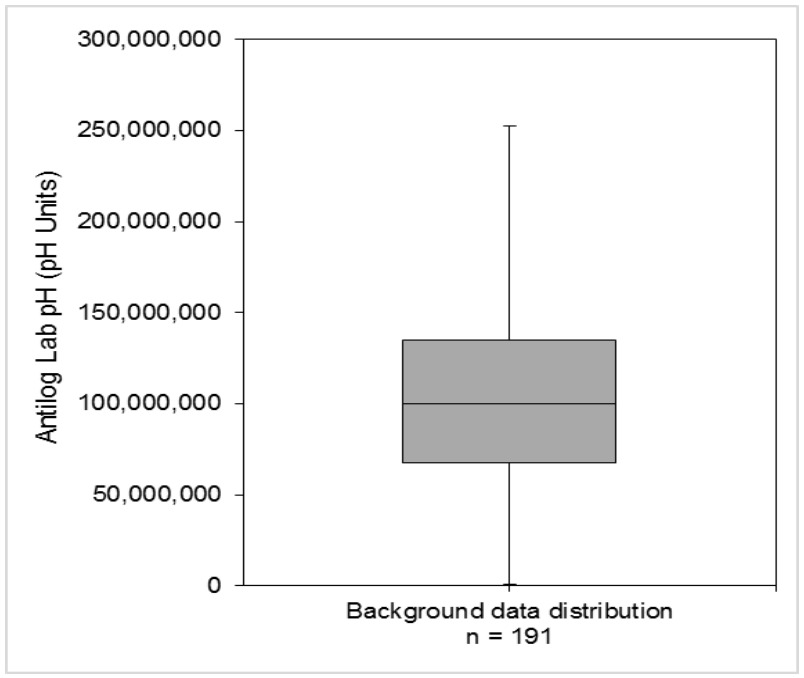
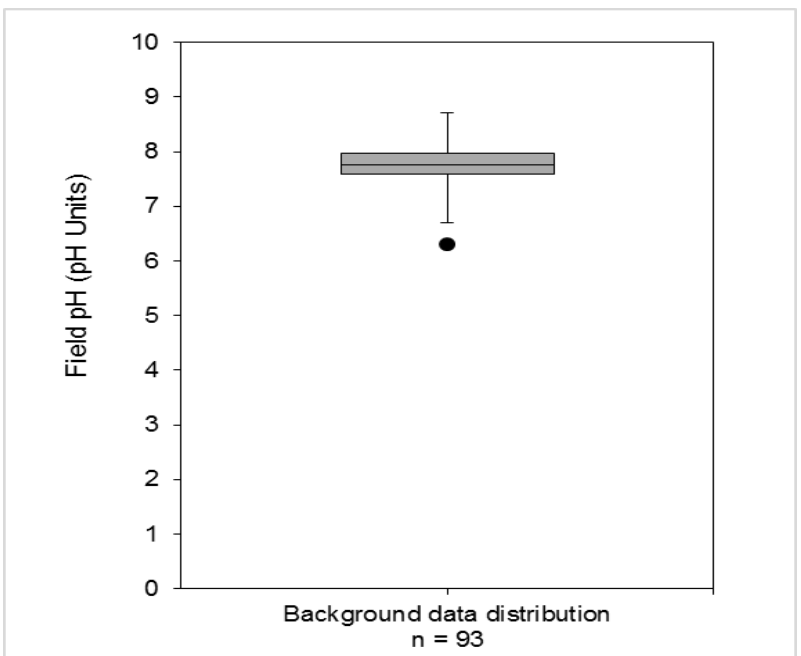


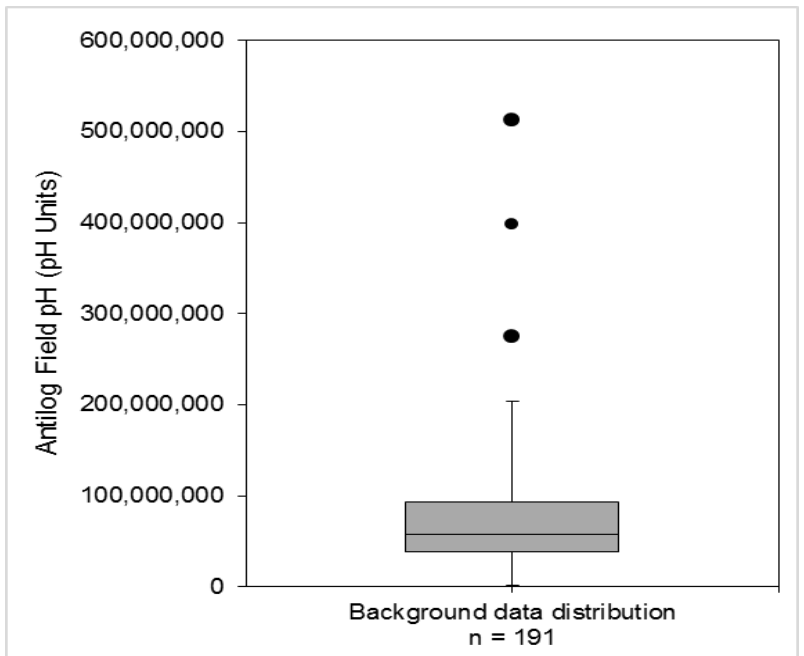
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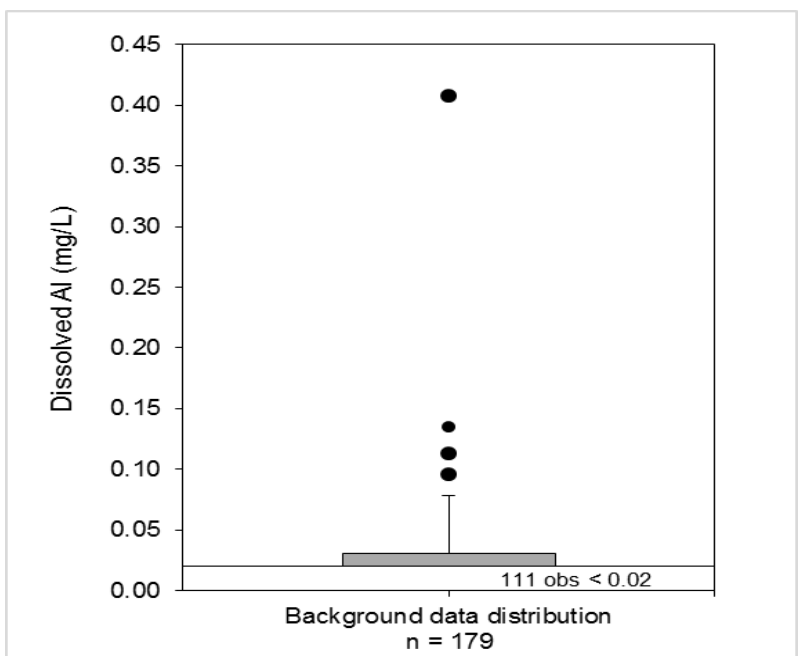
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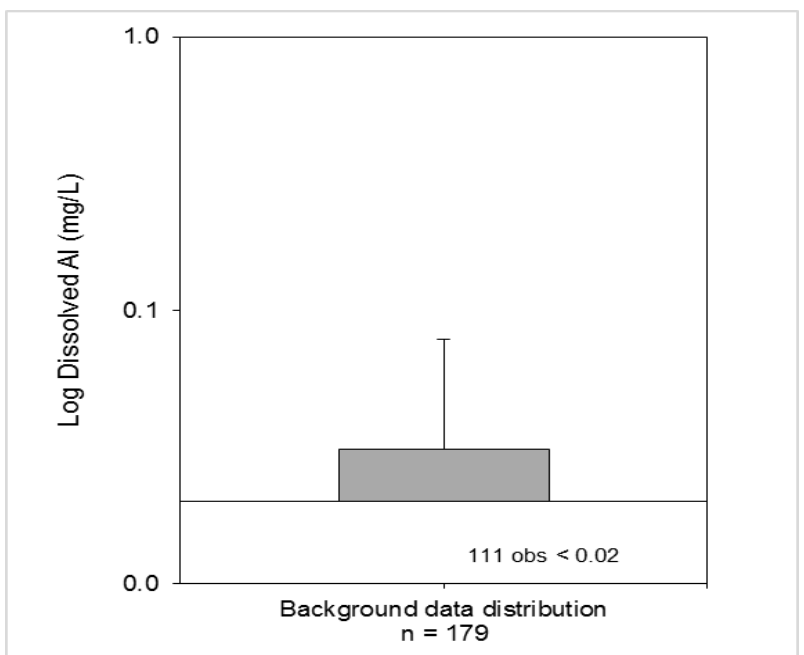


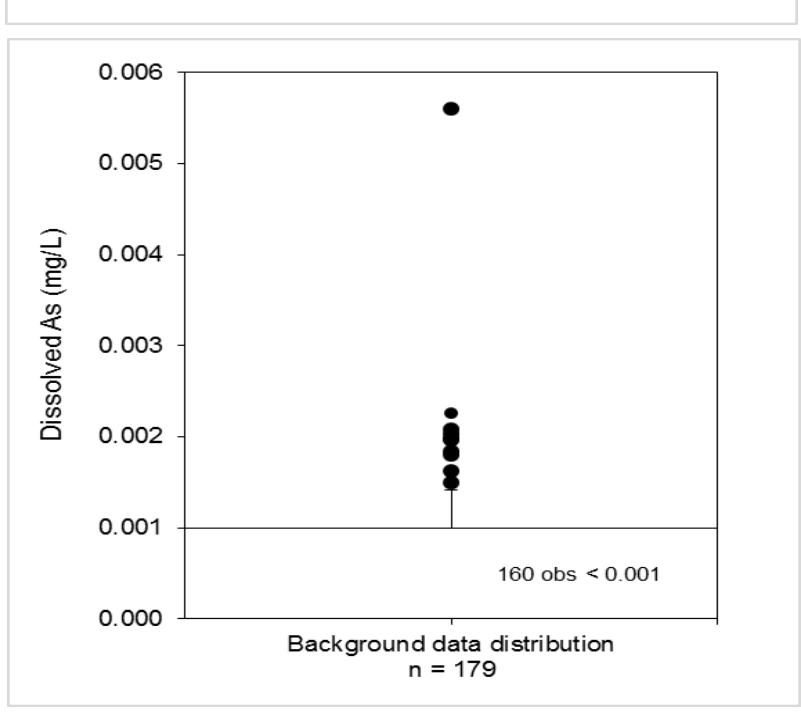












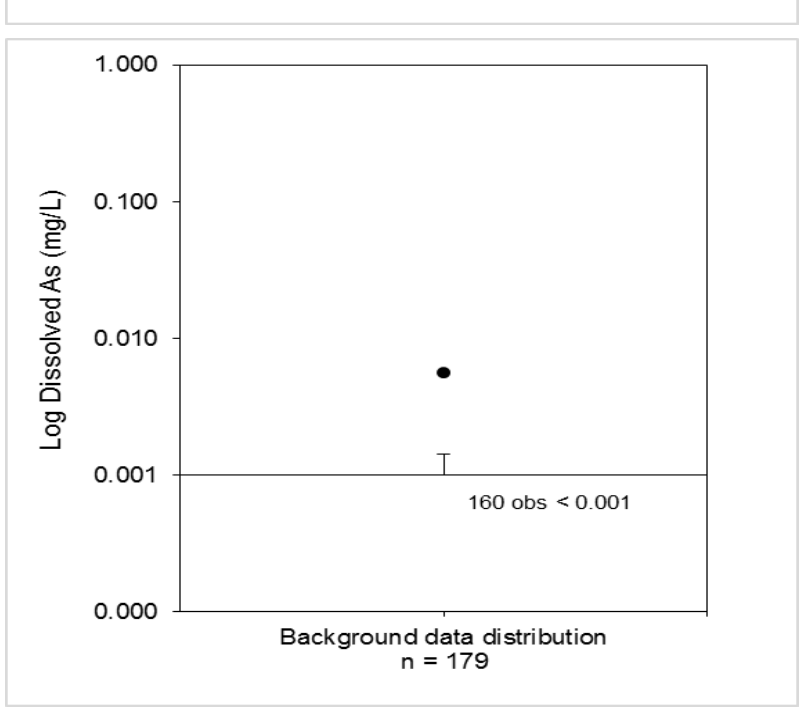
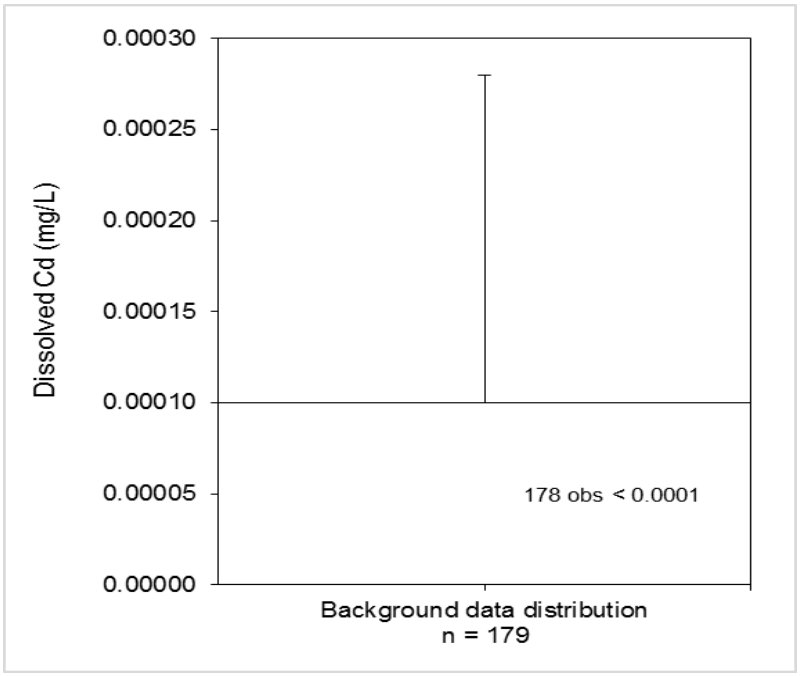
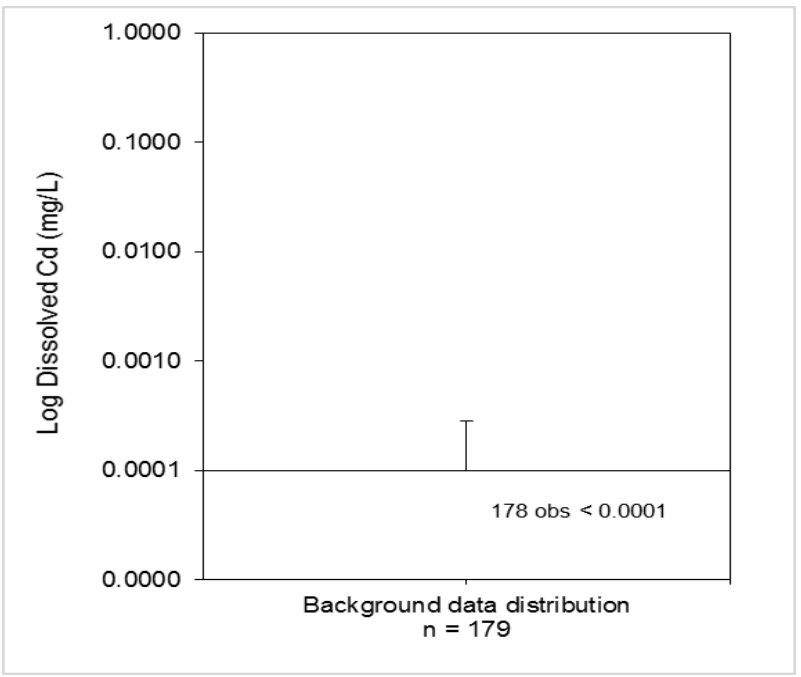
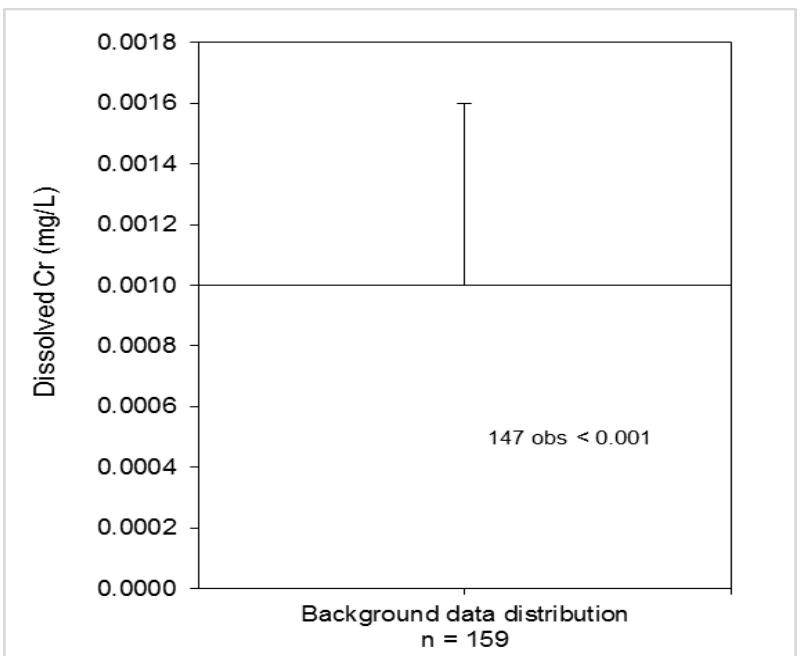


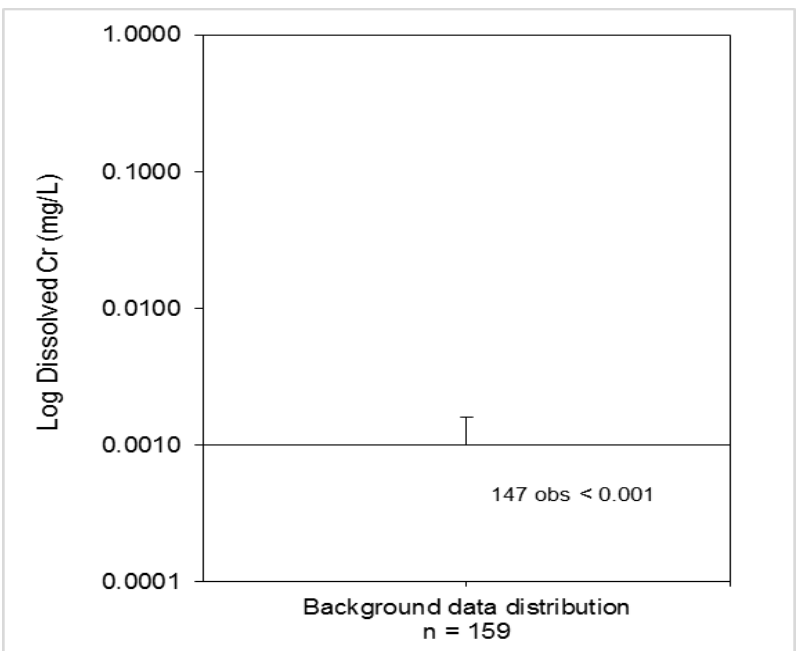
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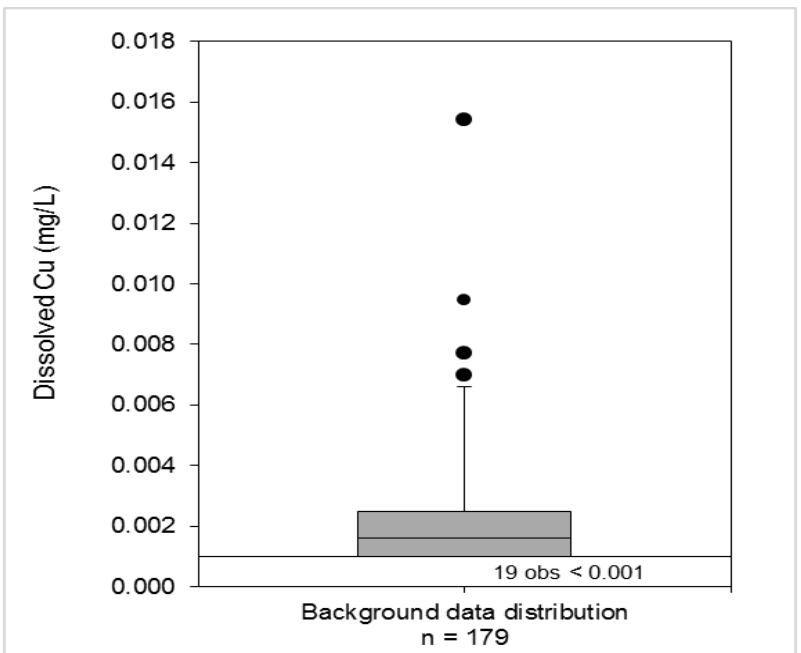
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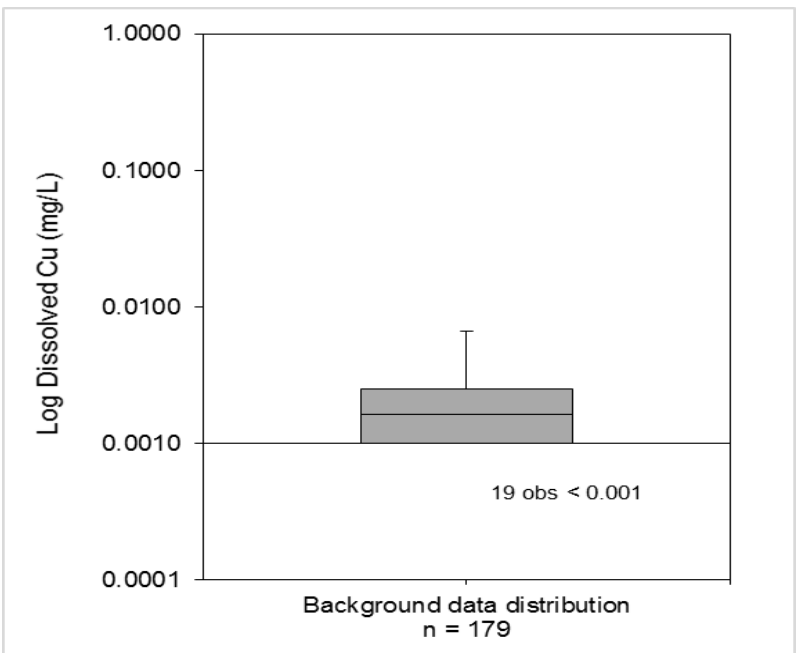


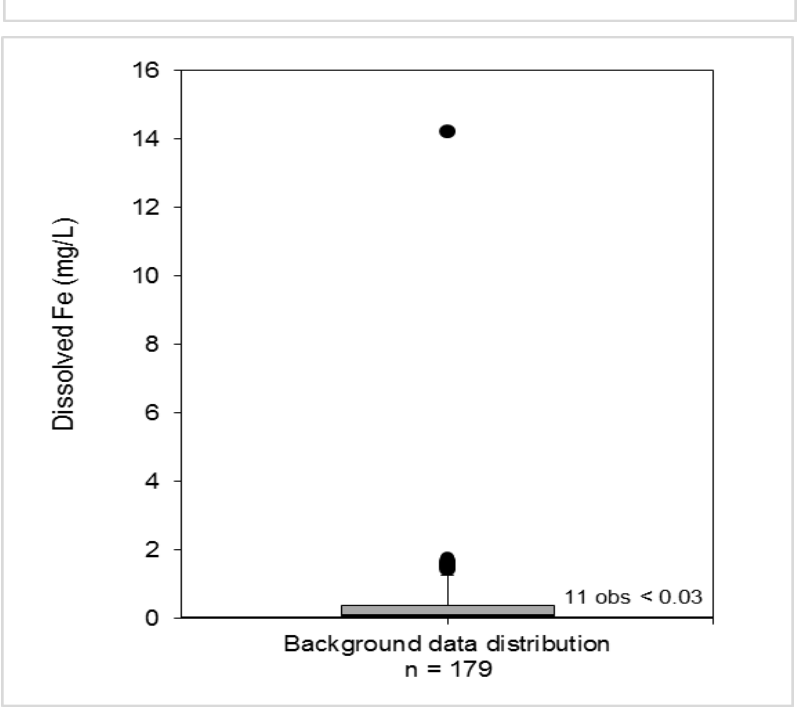












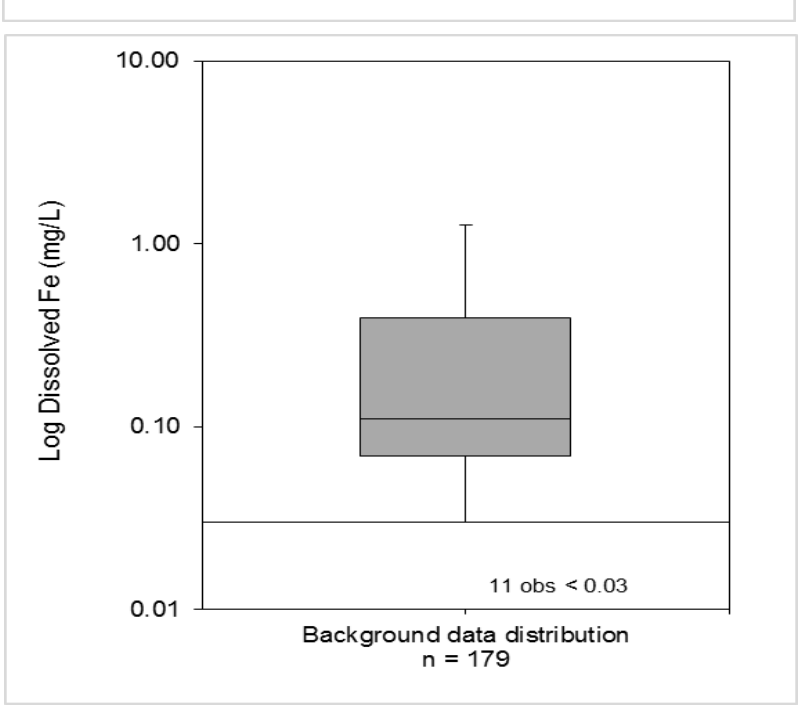
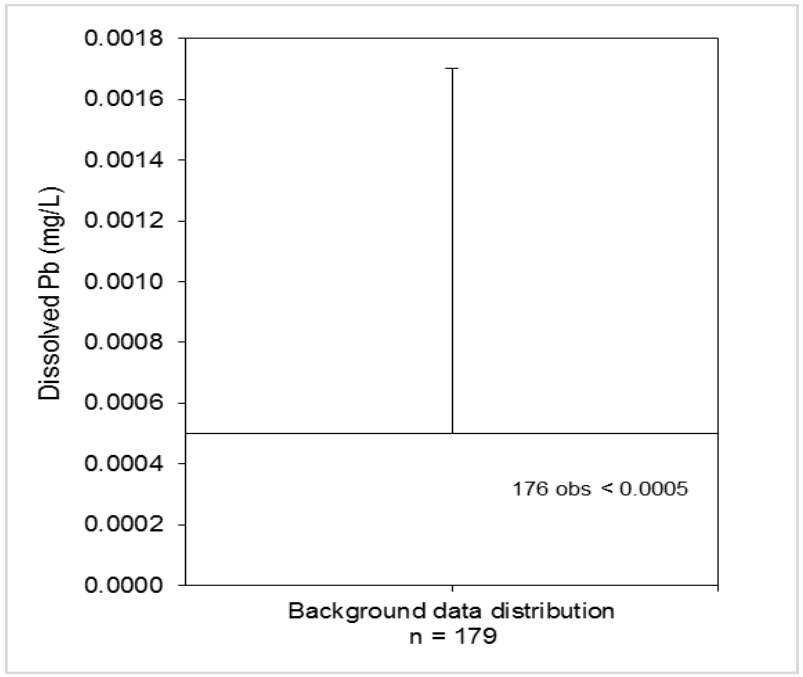
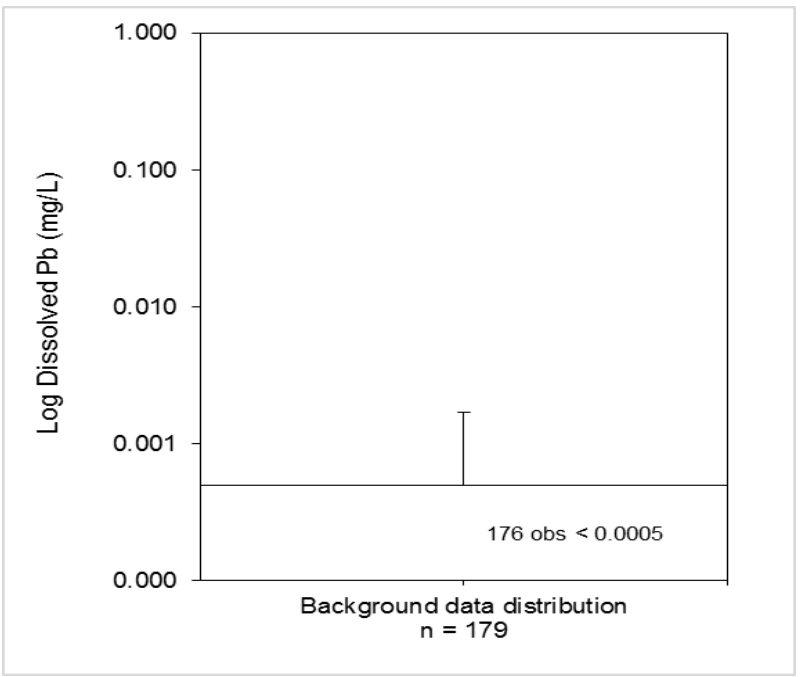
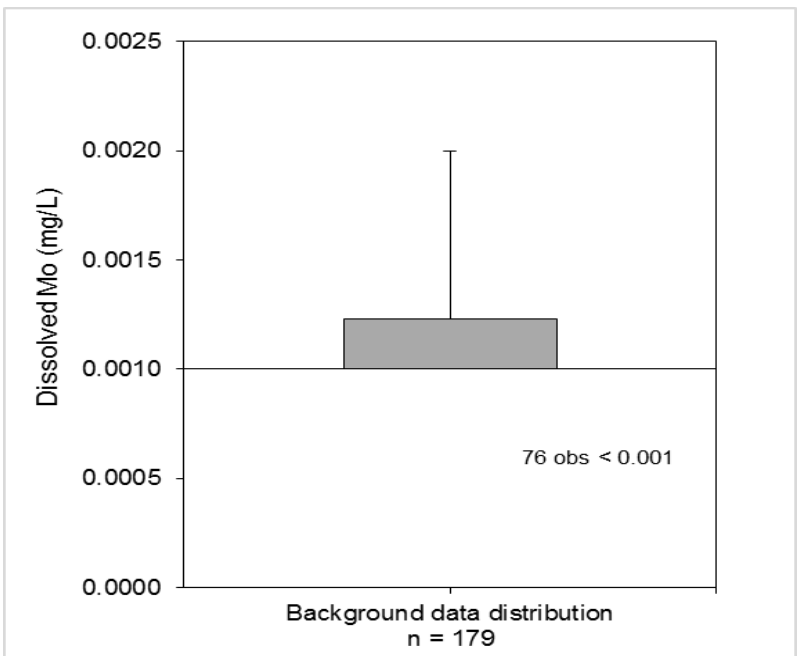


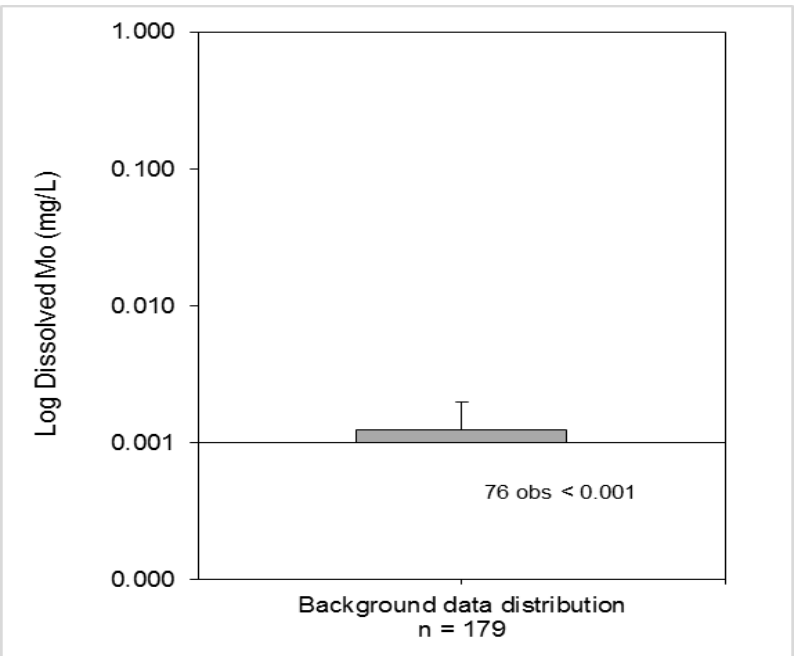
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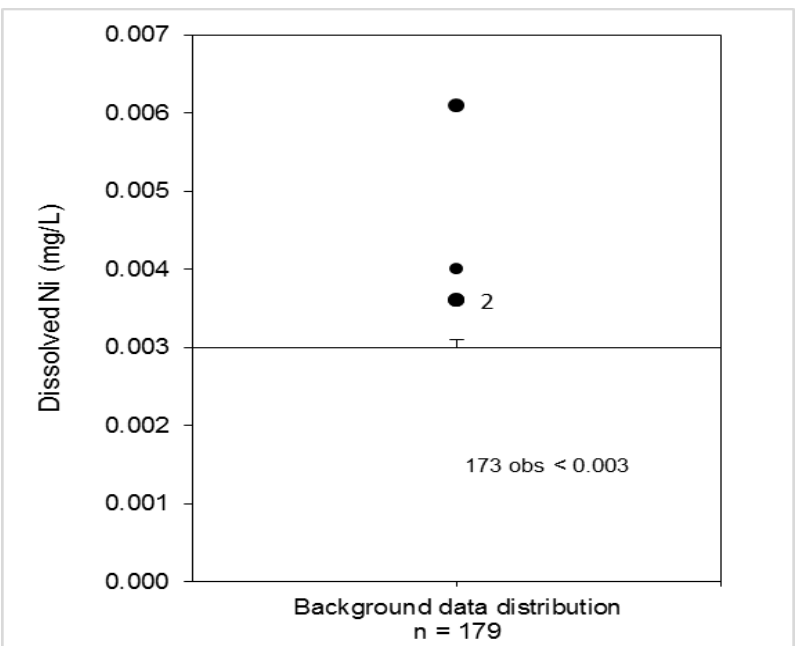
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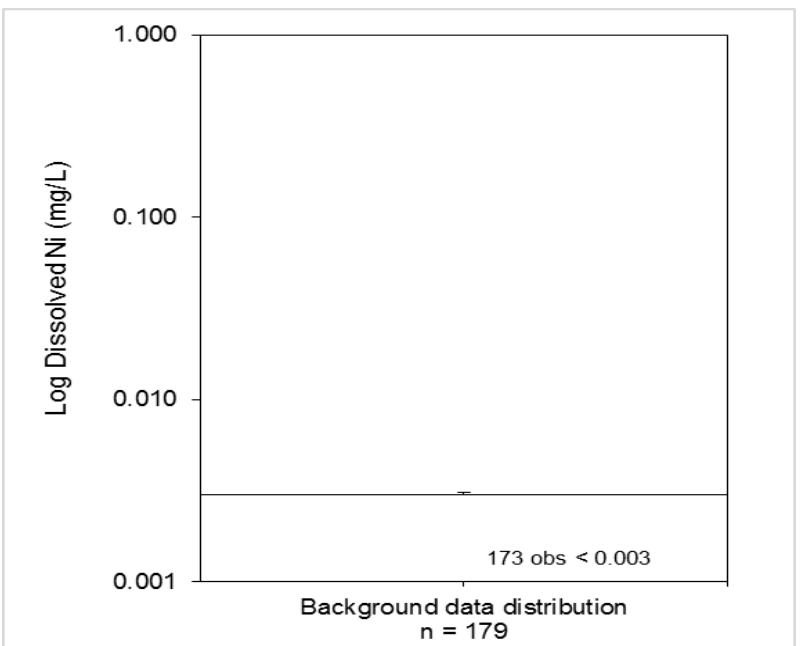


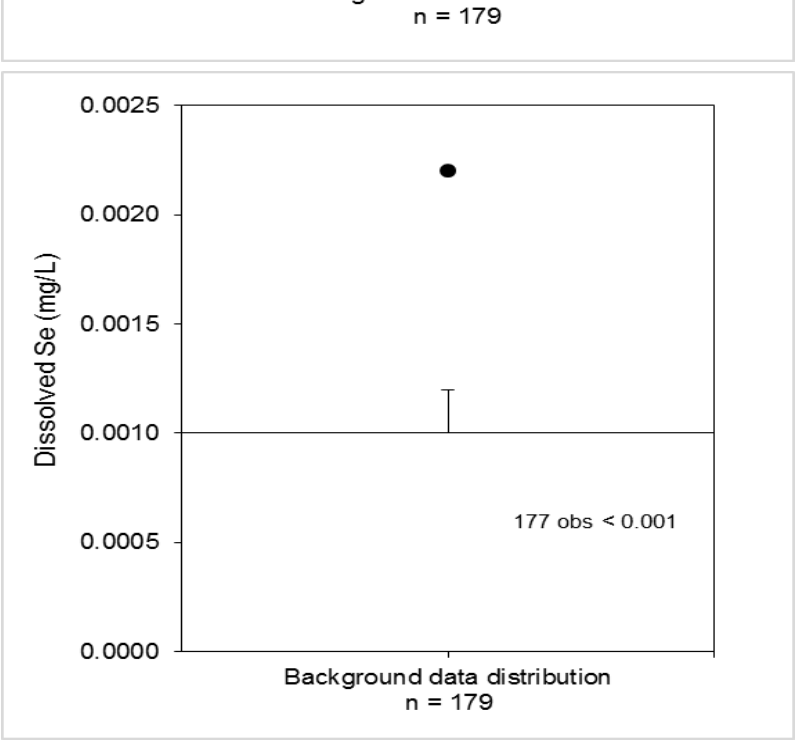












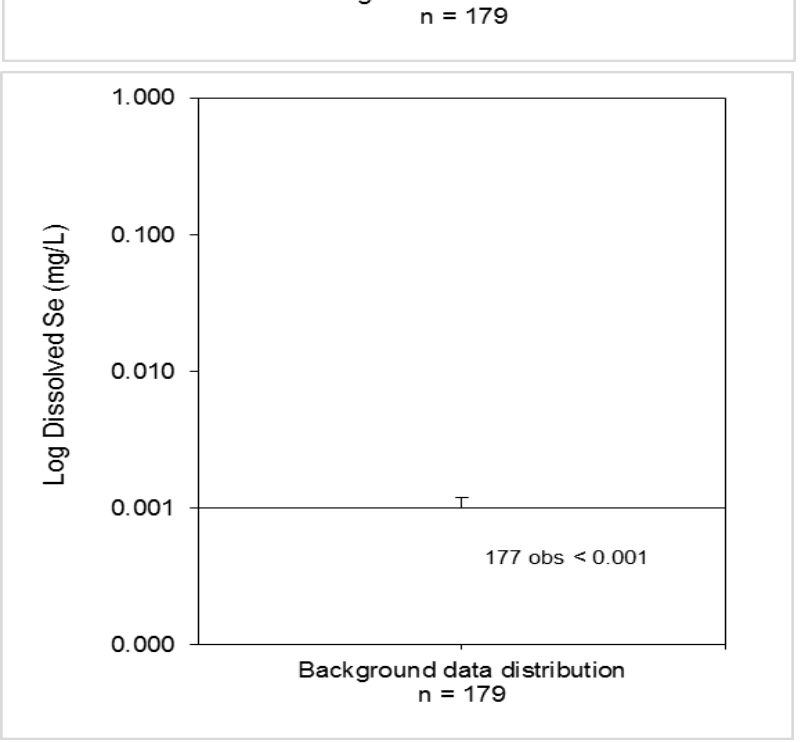
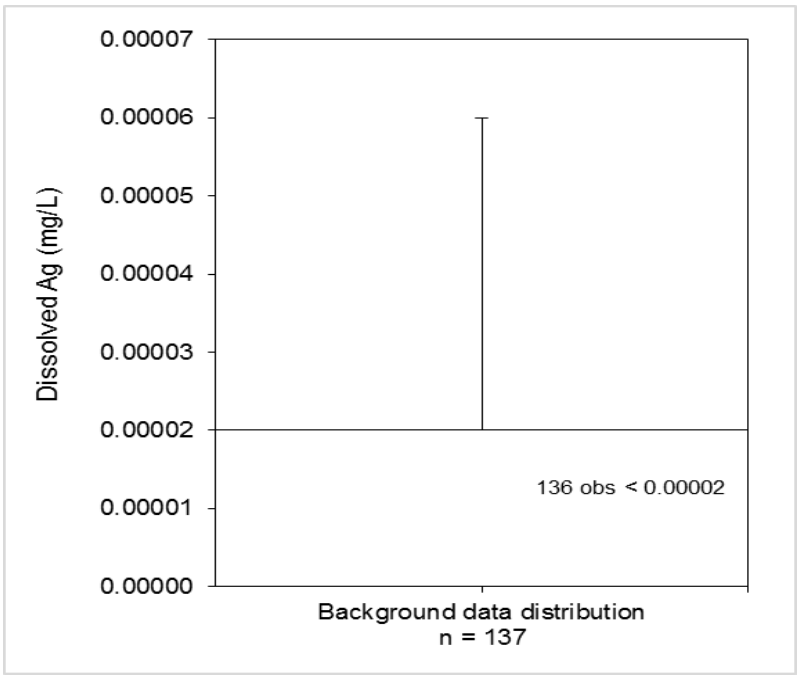
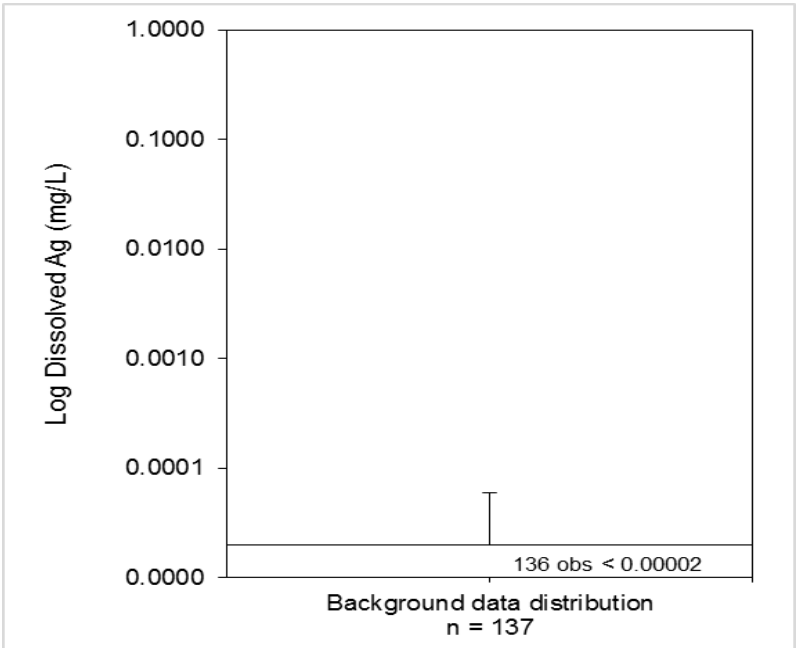
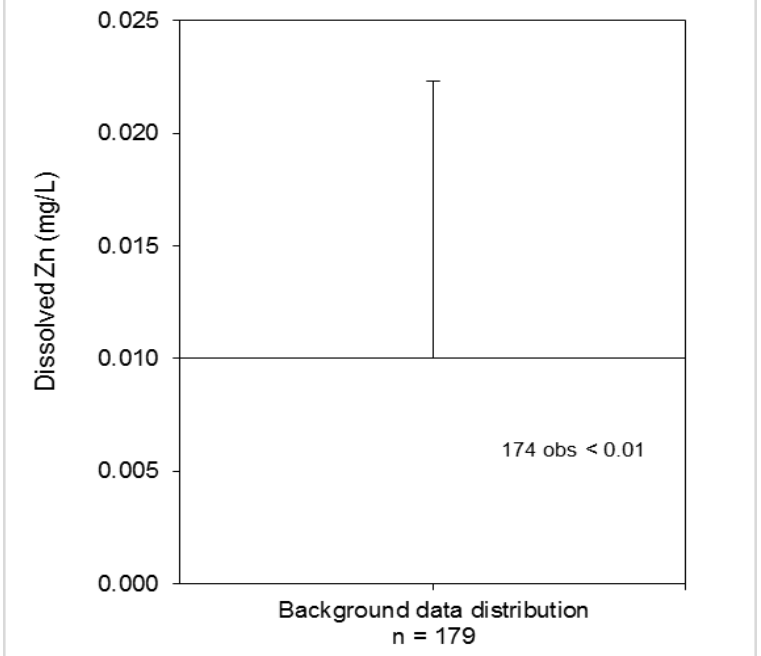


Figure C.1: Boxplots of analytes on non-transformed and log transformed scales. Black dots represent possible outliers. Line represents highest < MDL and any < MDL below this are represented as number of observations below this line.

NOTE: Possible outliers were only considered if they were detected on both non-transformed and log transformed scales







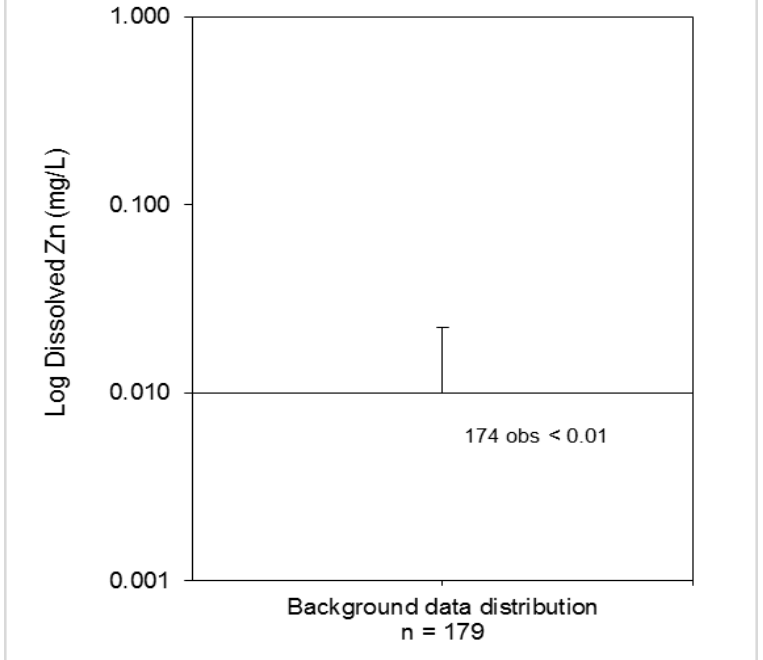


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APPENDIX D

WATER QUALITY DATA SUMMARY CALCULATIONS

Table D.1: Comparisons of dissolved arsenic ratios, relative to other analytes. Highlighted cells represent the sample with the potential arsenic concentration outlier

Station	Date	Arsenic Relative to:																				
		DOC (mg/L)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	TSS (mg/L)	TDS (mg/L)	Ammonia (mg/L)	Nitrite (mg/L)	Nitrate (mg/L)	Lab pH (pH Units)	Field pH (pH Units)	Dissolved Aluminum (mg/L)	Dissolved Cadmium (mg/L)	Dissolved Chromium (mg/L)	Dissolved Copper (mg/L)	Dissolved Iron (mg/L)	Dissolved Lead (mg/L)	Dissolved Molybdenum (mg/L)	Dissolved Nickel (mg/L)	Dissolved Selenium (mg/L)	Dissolved Silver (mg/L)	Dissolved Zinc (mg/L)
C4	11/17/2013	0.00018	0.000022	0.000021	0.00025	0.000021	0.020	0.22	0.056	0.00071	0.00078	0.17	254	4.3	5.9	0.00039	56	5.1	0.92	23	559	2.2
Summary Statistics	Average	0.00013	0.0000059	0.0000051	0.00010	0.0000038	0.034	0.32	0.026	0.000072	0.000093	0.038	76	1.2	0.40	0.0044	7.0	0.75	0.45	5.0	73	0.27
	Minimum	0.0000057	0.00000078	0.0000010	0.00000047	0.00000064	0.0012	0.0020	0.00067	0.000012	0.000013	0.0019	1.3	0.077	0.020	0.00039	0.18	0.067	0.10	0.045	6.7	0.011
	Maximum	0.0079	0.000024	0.000021	0.0011	0.000021	0.60	1.2	0.24	0.00071	0.00078	0.24	450	4.5	5.9	0.042	56	5.1	2.0	39	559	2.2
	Median	0.000047	0.0000046	0.0000044	0.000036	0.0000031	0.022	0.20	0.0083	0.000058	0.000064	0.033	60	1.0	0.25	0.0029	5.0	0.50	0.40	2.5	58	0.19
	95th Percentile	0.00010	0.000016	0.000010	0.00037	0.000010	0.081	1.1	0.10	0.00018	0.00025	0.084	282	3.0	1.0	0.013	19	1.7	0.83	18	180	0.84
	Standard Deviation	0.00076	0.0000043	0.0000030	0.00016	0.0000026	0.058	0.32	0.044	0.000068	0.000096	0.032	89	0.87	0.51	0.0052	6.6	0.66	0.24	6.7	60	0.29

Table D.2: Background data for Option 1 with median and 95th Percentile values.

Station	Date	DOC (mg/L)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	TSS (mg/L)	TDS (mg/L)	Ammonia (mg/L)	Nitrite (mg/L)	Nitrate (mg/L)	Lab pH (pH Units)	Field pH (pH Units)	Dissolved Aluminum (mg/L)	Dissolved Arsenic (mg/L)	Dissolved Cadmium (mg/L)	Dissolved Chromium (mg/L)	Dissolved Copper (mg/L)	Dissolved Iron (mg/L)	Dissolved Lead (mg/L)	Dissolved Molybdenum (mg/L)	Dissolved Nickel (mg/L)	Dissolved Selenium (mg/L)	Dissolved Silver (mg/L)	Dissolved Zinc (mg/L)
W1	5/27/2005			97	27	137	0.039	0.0020	0.054	8.05		0.018	0.00051	0.000025	0.00050	0.0034	0.17	0.00025	0.00050	0.0015	0.00050	0.000010	0.0025
W1	6/30/2005				1.5	189	0.010	0.00050	0.063	8.17		0.0065	0.00048	0.000025	0.00025	0.0020	0.046	0.000025	0.0011	0.00096	0.00050	0.0000050	0.0005
W1	7/28/2005			157	1.5	189	0.010	0.00050	0.058	8.03		0.0072	0.00048	0.000025	0.00025	0.0021	0.056	0.000073	0.0013	0.0011	0.00050	0.0000050	0.0025
W1	8/29/2005			138	4.0	186	0.041	0.00050	0.022	8.28		0.011	0.00060	0.0000085	0.00050	0.0024	0.25	0.00025	0.0010	0.0014	0.00050	0.000010	0.0025
W1	9/28/2005			134	10	178	0.026	0.00050	0.048	7.59		0.014	0.00058	0.0000085	0.00050	0.0026	0.33	0.00025	0.00050	0.0016	0.00050	0.000010	0.0025
W1	10/15/2005			140	15	181	0.020	0.00050	0.13	8.13		0.014	0.00054	0.000025	0.00025	0.0028	0.27	0.000025	0.00089	0.0014	0.00050	0.0000050	0.0014
W2	5/27/2005			86	50	109	0.018	0.0010	0.045	8.05		0.014	0.00025	0.000025	0.00050	0.0033	0.16	0.00025	0.00050	0.0015	0.00050	0.000010	0.0025
W2	6/30/2005				1.5	190	0.010	0.00050	0.062	8.17		0.0073	0.00044	0.000025	0.00025	0.0021	0.039	0.000025	0.0011	0.0010	0.00050	0.0000050	0.0015
W2	7/28/2005			153	1.5	194	0.010	0.00050	0.012	8.01		0.0071	0.00044	0.000025	0.00025	0.0021	0.05	0.000025	0.0013	0.0011	0.00050	0.0000050	0.0011
W2	8/30/2005			138	3.5	193	0.029	0.00050	0.014	8.26		0.012	0.00058	0.0000085	0.00050	0.0024	0.23	0.00025	0.0010	0.0015	0.00050	0.000010	0.0025
W2	9/28/2005			134	9.0	170	0.010	0.0010	0.046	7.90		0.014	0.00059	0.0000085	0.00050	0.0025	0.34	0.00025	0.00050	0.0017	0.00050	0.000010	0.0025
W2	10/15/2005			137	5.5	179	0.010	0.00050	0.13	8.11		0.013	0.00053	0.000025	0.00025	0.0027	0.25	0.000025	0.00086	0.0014	0.00050	0.0000050	0.00050
W3	5/27/2005			94	8.5	141	0.010	0.00050	0.012	7.99		0.027	0.00025	0.000025	0.00050	0.0066	0.12	0.00025	0.00050	0.0014	0.00050	0.000010	0.0025
W3	6/30/2005				1.5	224	0.010	0.00050	0.052	8.22		0.0080	0.00034	0.000025	0.00025	0.0034	0.042	0.00024	0.0016	0.00078	0.00050	0.0000050	0.012
W3	7/29/2005			166	26	198	0.010	0.00050	0.028	8.06		0.011	0.00041	0.000025	0.00025	0.0048	0.13	0.000025	0.0016	0.0012	0.00050	0.0000050	0.0015
W3	8/29/2005			135	7.5	182	0.042	0.00050	0.0025	8.24		0.018	0.00025	0.0000085	0.00050	0.0050	0.31	0.00025	0.0013	0.0014	0.00050	0.000010	0.0025
W3	9/28/2005			129	8.0	174	0.010	0.00050	0.013	7.99		0.026	0.00025	0.0000085	0.00050	0.0057	0.42	0.00025	0.0011	0.0014	0.00050	0.000010	0.0025
W3	10/15/2005			134	1.5	176	0.023	0.00050	0.039	8.08		0.022	0.00043	0.000025	0.00025	0.0061	0.39	0.000025	0.0012	0.0012	0.00050	0.0000050	0.0018
W9	5/27/2005			49	1.5	104	0.010	0.00050	0.0025	7.56		0.072	0.00025	0.000025	0.00050	0.0035	0.25	0.00025	0.00050	0.0013	0.00050	0.000010	0.0025
W9	6/30/2005				1.5	126	0.020	0.00050	0.0025	7.63		0.035	0.00054	0.000025	0.00025	0.0018	0.53	0.000025	0.00037	0.0012	0.00050	0.0000050	0.00050
W9	7/29/2005			95	3.7	129	0.010	0.00050	0.0025	7.51		0.030	0.00057	0.000025	0.00025	0.0033	0.65	0.00019	0.0011	0.0013	0.00050	0.0000050	0.0040
W9	8/29/2005			74	1.5	126	0.026	0.0016	0.0025	7.84		0.045	0.00058	0.000022	0.00050	0.0017	0.67	0.00025	0.00050	0.0012	0.00050	0.000010	0.0025
W9	9/28/2005			68	1.5	118	0.010	0.00050	0.0025	7.62		0.058	0.00025	0.0000085	0.00050	0.0022	0.60	0.00025	0.00050	0.0011	0.00050	0.000010	0.0025
W9	10/15/2005			64	1.5	111	0.010	0.00050	0.0025	7.19		0.056	0.00039	0.000025	0.00025	0.0021	0.44	0.000025	0.00031	0.00096	0.00050	0.0000050	0.0017
W10	5/27/2005			46	1.5	115	0.019	0.00050	0.0094	7.47		0.14	0.00025	0.000025	0.00050	0.0095	0.061	0.00025	0.00050	0.0011	0.00050	0.000010	0.0055
W10	6/30/2005				6.0	116	0.036	0.00050	0.0025	7.31		0.11	0.00059	0.000025	0.00062	0.0047	0.40	0.000082	0.00014	0.0018	0.00050	0.0000050	0.022
W10	7/29/2005			55	4.2	105	0.010	0.00050	0.0025	7.34		0.068	0.00037	0.000025	0.00025	0.0044	0.099	0.00011	0.00014	0.0012	0.00050	0.0000050	0.0074
W10	8/29/2005			70	13	117	0.033	0.00050	0.0025	7.74		0.043	0.00025	0.0000085	0.00050	0.0040	0.081	0.00025	0.00050	0.0016	0.00050	0.000010	0.0052
W10	9/28/2005			101	15	141	0.010	0.00050	0.0025	7.77		0.030	0.00025	0.0000085	0.00050	0.0033	0.048	0.00025	0.00050	0.00050	0.00050	0.000010	0.0025
W10	10/15/2005			100	6.5	138	0.010	0.00050	0.0025	7.30		0.036	0.00030	0.000025	0.00025	0.0041	0.055	0.00003	0.00024	0.00073	0.00050	0.0000050	0.0015
W6	5/27/2005			65	14	103	0.010	0.0016	0.032	7.91		0.014	0.00025	0.000025	0.00050	0.0016	0.072	0.00025	0.00050	0.0015	0.00050	0.000010	0.0025
W6	6/30/2005				12	123	0.010	0.00050	0.0086	7.41		0.0056	0.00039	0.000025	0.00025	0.0010	0.015	0.000025	0.00045	0.00099	0.00050	0.0000050	0.021
W6	7/29/2005			171	22	198	0.010	0.00050	0.022	8.01		0.0070	0.00026	0.000025	0.00025	0.0025	0.062	0.000083	0.00082	0.00060	0.00050	0.0000050	0.0025
W6	8/30/2005			106	1.5	147	0.010	0.00050	0.0025	8.17		0.0067	0.00025	0.0000085	0.00050	0.0010	0.064	0.00025	0.00050	0.0011	0.00050	0.000010	0.0025
W6	9/28/2005			106	3.5	133	0.010	0.00050	0.0025	7.95		0.0067	0.00025	0.0000085	0.00050	0.0011	0.10	0.00025	0.00050	0.0012	0.00050	0.000010	0.0025
W6	10/15/2005			98	1.5	128	0.010	0.00050	0.0061	7.92		0.077	0.00042	0.000025	0.00025	0.0011	0.088	0.000025	0.00040	0.00118	0.00050	0.0000050	0.0015
W6	6/2/2006				115	96	0.021	0.00050	0.024	7.64		0.025	0.00044	0.000025	0.00025	0.0018	0.096	0.000025	0.00031	0.00182	0.00050	0.0000050	0.0030
W6	8/25/2006		120			160			0.015	8.01													
W6	9/6/2006		110		4.0				0.015	7.86		0.0080	0.00050	0.0000050	0.00060	0.0010	0.11	0.000050	0.00050	0.00060	0.00010		0.0020
W6	6/20/2007			81	16	130	0.025	0.025	0.10	7.77													
W6	9/18/2007			93	8.0	172	0.025	0.010	0.010	7.95													
W6	6/17/2008			86	17	120	0.025	0.070	0.010	7.82													
W6	9/18/2008	18			4.0	150	0.025	0.0050	0.0050	7.80		0.012	0.00040	0.0000050	0.0013	0.0020	0.090	0.000050	0.00027	0.0020	0.0012	0.0000050	0.0030
W6	7/17/2012	8.0	104	163	1130	122	0.14	0.0025	0.026	7.98	7.92	0.016	0.00053	0.0000050	0.00050	0.0011	0.068	0.00010	0.00050	0.0010	0.00012	0.000010	0.0025
W6	7/30/2012	8.1	101	145	614	124	0.070	0.0072	0.052	8.09	7.89	0.019	0.00059	0.0000050	0.00050	0.00089	0.081	0.00010	0.00050	0.00050	0.000050	0.000010	0.0025
W6	8/9/2012		112	136	566	144	0.025	0.0066	0.022	8.17	8.09	0.017	0.00081	0.0000050	0.00050	0.00083	0.085	0.00010	0.00050	0.0010	0.00010	0.000010	0.0025
W6	8/25/2012	7.7	103	161	1160	138	0.023	0.0025	0.045	8.00	7.76	0.014	0.00081	0.0000050	0.00050	0.0010	0.080	0.00010	0.00050	0.0010	0.000050	0.000010	0.0025
W6	9/13/2012	14	88	93	50	118	0.011	0.0025	0.034	8.04	7.95	0.017	0.00047	0.0000050	0.00050	0.0015	0.11	0.00010	0.00050	0.0014	0.000050	0.000010	0.0025
W6	10/13/2012	10		117	45	150	0.016	0.0025	0.042	7.92	8.30												

Table D.2: Background data for Option 1 with median and 95th Percentile values.

Station	Date	DOC (mg/L)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	TSS (mg/L)	TDS (mg/L)	Ammonia (mg/L)	Nitrite (mg/L)	Nitrate (mg/L)	Lab pH (pH Units)	Field pH (pH Units)	Dissolved Aluminum (mg/L)	Dissolved Arsenic (mg/L)	Dissolved Cadmium (mg/L)	Dissolved Chromium (mg/L)	Dissolved Copper (mg/L)	Dissolved Iron (mg/L)	Dissolved Lead (mg/L)	Dissolved Molybdenum (mg/L)	Dissolved Nickel (mg/L)	Dissolved Selenium (mg/L)	Dissolved Silver (mg/L)	Dissolved Zinc (mg/L)
W6	11/12/2012	8.7	153	155	2.2	162	0.025	0.0025	0.010	7.93	6.69	0.0062	0.00039	0.0000050	0.00050	0.0014	0.024	0.00010	0.00050	0.0014	0.00010	0.000010	0.0025
W6	5/14/2013	22	31	34	13	84	0.13	0.0025	0.010	7.48	7.60	0.059	0.00037	0.000012	0.00050	0.0035	0.18	0.00010	0.00050	0.0013	0.000050	0.000010	0.0025
W6	6/22/2013	7.8	95	115	172	118	0.024	0.013	0.028	8.07	7.61	0.0095	0.00053	0.0000050	0.00050	0.00096	0.072	0.00010	0.00050	0.0012	0.000050	0.000010	0.0025
W6	7/12/2013	8.3	88	95	25	136	0.012	0.0025	0.010	7.90	7.69	0.0066	0.00046	0.0000050	0.00050	0.00095	0.11	0.00010	0.00050	0.0011	0.000050	0.000010	0.0025
W6	8/22/2013	7.4	102	106	20	144	0.014	0.0025	0.010	8.04	7.81	0.0084	0.00052	0.0000050	0.00050	0.0012	0.10	0.00010	0.00050	0.0011	0.000050	0.000010	0.0025
W6	9/6/2013	6.4	101	114	5.2	130	0.0061	0.0025	0.030	7.95	7.60	0.0083	0.00056	0.0000050	0.00050	0.00089	0.11	0.00010	0.00050	0.0011	0.000050	0.000010	0.0025
W6	10/7/2013	8.2	107	106	74	176	0.044	0.0025	0.042	8.08	7.66	0.013	0.00051	0.0000050	0.00050	0.00097	0.15	0.00010	0.00050	0.0013	0.000050	0.000010	0.0025
W6	11/16/2013	8.8	119	123	33	120	0.015	0.0025	0.010	8.11	7.96	0.0057	0.00048	0.0000050	0.00050	0.00084	0.12	0.00010	0.00050	0.0010	0.000050	0.000010	0.0025
W6	12/7/2013	8.8	159	182	3.6	194	0.0076	0.0025	0.010	8.21	7.50	0.0041	0.00061	0.000011	0.00050	0.0016	0.014	0.00010	0.00050	0.0011	0.000050	0.000010	0.0025
W7	5/27/2005			77	31	116	0.010	0.0018	0.082	7.95		0.024	0.00025	0.000025	0.00050	0.0018	0.14	0.00025	0.00050	0.0016	0.00050	0.000010	0.0025
W7	6/30/2005				14	153	0.010	0.0017	0.17	7.87		0.0086	0.00042	0.000025	0.00025	0.0012	0.070	0.000025	0.0012	0.00093	0.00050	0.0000050	0.0064
W7	7/29/2005			167	20	197	0.010	0.00050	0.17	7.18		0.0096	0.00035	0.000025	0.00025	0.0040	0.11	0.000025	0.0013	0.00095	0.00050	0.0000050	0.0020
W7	8/30/2005			109	3.0	148	0.020	0.00050	0.023	8.19		0.016	0.00051	0.0000085	0.00050	0.0014	0.29	0.00025	0.0010	0.0015	0.00050	0.000010	0.0025
W7	9/28/2005			107	361	142	0.010	0.00050	0.059	8.02		0.019	0.00054	0.0000085	0.00050	0.0016	0.32	0.00025	0.00050	0.0018	0.00050	0.000010	0.0025
W7	10/15/2005			106	7.5	145	0.021	0.00050	0.10	6.75		0.016	0.00048	0.000025	0.00025	0.0013	0.25	0.000025	0.00093	0.0015	0.00050	0.0000050	0.00050
W7	6/2/2006				705	104	0.062	0.0015	0.045	7.61		0.071	0.00062	0.000025	0.00055	0.0021	0.55	0.000054	0.00070	0.0022	0.00050	0.0000050	0.00050
W7	6/8/2006																						
W7	6/15/2006		100		53	114	0.027		0.18	7.86		0.013	0.00040	0.0000050	0.00025	0.0010	0.090	0.000050	0.0010	0.0010	0.00020		0.00050
W7	6/23/2006		122		328	176	0.028		0.23	7.78		0.011	0.00050	0.0000050	0.00070	0.00050	0.090	0.000050	0.0010	0.0013	0.00030		0.0020
W7	6/28/2006		107		103	120	0.010		0.11	7.90		0.41	0.00080	0.000030	0.0015	0.0030	0.060	0.00050	0.00050	0.0036	0.00010		0.0050
W7	7/7/2006		110		3.0	184			0.21	7.91		0.0025	0.00060	0.0000050	0.0011	0.0010	0.060	0.000050	0.0020	0.0015	0.00010		0.0010
W7	7/12/2006		110		1.0		0.014		0.18	7.77		0.0060	0.00050	0.0000050	0.0010	0.0010	0.060	0.000050	0.0010	0.00070	0.00020		0.00050
W7	7/20/2006		130		2.0		0.012		0.19	7.88		0.0060	0.00040	0.0000050	0.00070	0.00050	0.11	0.00020	0.0010	0.00060	0.00010		0.00050
W7	7/26/2006		128		9.0		0.016		0.18	7.87		0.0080	0.00040	0.0000050	0.00025	0.0020	0.10	0.000050	0.00050	0.0021	0.00010		0.00050
W7	8/2/2006		128		4.0		0.015		0.15	7.79		0.0080	0.00010	0.0000050	0.00025	0.0010	0.070	0.000050	0.00050	0.00060	0.00010		0.00050
W7	8/10/2006		126		1.0		0.014		0.16	7.80		0.0090	0.00050	0.0000050	0.00070	0.0010	0.080	0.000050	0.0010	0.00060	0.00010		0.0010
W7	8/25/2006		129			180	0.027		0.12	7.92		0.012	0.00040	0.0000050	0.00060	0.0010	0.080	0.000050	0.0010	0.00090	0.00010		0.0030
W7	8/30/2006		132		1.0	200	0.012		0.11	7.79		0.013	0.00040	0.0000050	0.00080	0.0020	0.10	0.000050	0.0010	0.0012	0.00010		0.0050
W7	9/6/2006		130		4.0		0.011		0.080	7.88		0.0050	0.00050	0.0000050	0.00080	0.0020	0.10	0.000050	0.0010	0.0010	0.00010		0.0020
W7	9/13/2006		130		5.0		0.011		0.094	7.84		0.0080	0.00030	0.0000050	0.00090	0.0010	0.11	0.000050	0.0010	0.00090	0.00010		0.0010
W7	9/20/2006		132		3.0		0.0080		0.11	7.85		0.0070	0.00050	0.00028	0.00070	0.0010	0.090	0.000050	0.0010	0.00025	0.00020		0.00050
W7	9/28/2006		129			182	0.011		0.080	7.26		0.011	0.00040	0.0000050	0.00080	0.0010	0.080	0.000050	0.0010	0.00080	0.00010		0.0020
W7	10/4/2006		126		2.0		0.079		0.10	7.94		0.0025	0.00030	0.0000050	0.00025	0.00050	0.11	0.000050	0.0010	0.00070	0.00010		0.0010
W7	10/12/2006		132		4.0	164	0.024		0.12	7.39		0.0070	0.00040	0.0000050	0.00025	0.0010	0.090	0.000050	0.0010	0.00060	0.00030		0.0020
W7	6/5/2007			113	75	162	0.025	0.033	0.050	7.97													
W7	6/20/2007	8.2	110		6.0	164	0.025	0.025	0.30	7.84		0.014	0.00040	0.0000050	0.00060	0.0020	0.050	0.000050	0.0010	0.00025	0.00010		0.0040
W7	7/18/2007	14	128	125	5.7	179	0.025	0.025	0.30	7.78		0.011	0.00050	0.0000050	0.0010	0.0020	0.040	0.000050	0.0010	0.00070	0.00010		0.0060
W7	8/24/2007			124	4.7	192	0.025	0.025	0.20	8.04													
W7	9/18/2007	11	110		20	182	0.025	0.010	0.050	8.00		0.0070	0.00040	0.0000050	0.0016	0.0010	0.080	0.000050	0.0020	0.00080	0.00010		0.0020
W7	10/30/2007			129	6.5	174	0.025	0.14	0.010	7.73													
W7	4/22/2008		187		1.0	268	0.025	0.055	0.010	8.40		0.010	0.00027	0.000040	0.00097	0.0012	0.030	0.000067	0.0017	0.0010	0.00030		0.0017
W7	6/3/2008		92	98	15	137	0.025	0.040	0.037	8.02		0.0050	0.00010	0.000040	0.0013	0.00050	0.050	0.000050	0.0012	0.0010	0.00030	0.0000050	0.0020
W7	6/17/2008				10	156	0.025	0.050	0.14	7.95		0.020	0.00010	0.000040	0.0013	0.0010	0.070	0.000050	0.0015	0.0010	0.00220	0.0000050	0.0020
W7	8/6/2008	9.0	123	124	105	167	0.025	0.0050	0.14	7.92		0.014	0.00040	0.0000050	0.00060	0.0020	0.070	0.000050	0.0010	0.0010	0.00010	0.000060	0.0060
W7	9/18/2008	17			28	170	0.025	0.0050	0.030	7.89		0.070	0.00030	0.0000050	0.0016	0.0040	0.20	0.0017	0.00072	0.0010	0.00030	0.0000050	0.0030
W7	10/28/2008	11	123		5.7	183	0.025			7.84		0.0090	0.00050	0.0000050	0.00050	0.0010	0.078	0.000050	0.0011	0.0010	0.00030	0.0000050	0.0010
W7	4/30/2009	18	20		1.0	52				6.84	7.60	0.053	0.00010	0.000080	0.00060	0.0050	0.070	0.00010	0.00013	0.00050	0.00030	0.0000050	0.0090
W7	6/8/2009	9.2	93		32	156				7.90	7.74	0.013	0.00030	0.0000050	0.00020	0.0020	0.050	0.000050	0.0013	0.0010	0.00030	0.0000050	0.0020
W7	7/28/2009		137		4.0	184				8.07													

Table D.2: Background data for Option 1 with median and 95th Percentile values.

Station	Date	DOC (mg/L)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	TSS (mg/L)	TDS (mg/L)	Ammonia (mg/L)	Nitrite (mg/L)	Nitrate (mg/L)	Lab pH (pH Units)	Field pH (pH Units)	Dissolved Aluminum (mg/L)	Dissolved Arsenic (mg/L)	Dissolved Cadmium (mg/L)	Dissolved Chromium (mg/L)	Dissolved Copper (mg/L)	Dissolved Iron (mg/L)	Dissolved Lead (mg/L)	Dissolved Molybdenum (mg/L)	Dissolved Nickel (mg/L)	Dissolved Selenium (mg/L)	Dissolved Silver (mg/L)	Dissolved Zinc (mg/L)
W7	8/6/2009		121		2.0	202			0.18	8.35	7.50												
W7	8/25/2009	8.2									8.04												
W7	8/31/2009		110									0.010	0.00040	0.0000050	0.0012	0.0040	0.10	0.00040	0.0011	0.0010	0.00030	0.0000050	0.0010
W7	10/7/2009	7.8	125	141	3.7	140	0.010	0.0025	0.14	7.99	8.24	0.0070	0.00030	0.000040	0.00050	0.0012	0.046	0.00010	0.0010	0.00050	0.00020	0.000010	0.0025
W7	10/25/2009	8.7	126	141	2.0	120	0.010	0.0025	0.15	8.05	8.23	0.0070	0.00030	0.000040	0.00050	0.0012	0.046	0.00010	0.0010	0.00050	0.00020	0.000010	0.0025
W7	4/19/2010			80	3.5	142				7.81	7.31	0.019	0.00030	0.000020	0.00040	0.0060	0.090	0.00030	0.00080	0.00050	0.00030	0.0000050	0.0050
W7	4/27/2010	23	36	35	2.0	62	0.015	0.0025	0.010	7.60	7.65	0.075	0.00030	0.0000050	0.00050	0.0041	0.15	0.00010	0.00050	0.0010	0.000050	0.000010	0.0025
W7	4/29/2010	19	36	36	2.0	62	0.016	0.0050	0.020	7.50	7.72	0.065	0.00030	0.0000050	0.00050	0.0034	0.18	0.00010	0.00050	0.0010	0.000050	0.000010	0.0025
W7	5/1/2010	19	41	44	15	74	0.031	0.0080	0.070	7.70	7.15	0.053	0.00040	0.000020	0.00050	0.0032	0.26	0.00010	0.00050	0.0010	0.000050	0.000010	0.0025
W7	5/3/2010	22	45	46	12	80	0.016	0.0025	0.010	7.70	7.74	0.047	0.00040	0.0000050	0.00050	0.0027	0.38	0.00010	0.00050	0.0010	0.000050	0.000010	0.0025
W7	5/5/2010	15	49	55	12	76	0.0050	0.0025	0.010	7.80		0.038	0.00050	0.0000050	0.00050	0.0025	0.57	0.00010	0.00050	0.0010	0.000050	0.000010	0.0025
W7	5/7/2010	13	57	56	18	62	0.051	0.0070	0.10	7.80	7.95	0.028	0.00040	0.000010	0.00050	0.0026	0.36	0.00010	0.00050	0.0020	0.000050	0.000010	0.0025
W7	5/9/2010	13	80	71	2.0	130	0.0025	0.0025	0.070	8.00	8.03	0.023	0.00040	0.0000050	0.00050	0.0018	0.30	0.00010	0.00050	0.0010	0.00010	0.000010	0.0025
W7	5/10/2010	12	81	90	2.0	130	0.015	0.0060	0.13	8.10	8.06	0.021	0.00040	0.000020	0.00050	0.0022	0.28	0.00010	0.00050	0.0010	0.00010	0.000010	0.0025
W7	5/12/2010	10	82	85	2.0	100	0.025	0.0025	0.030	8.00		0.017	0.00040	0.0000050	0.00050	0.0016	0.17	0.00010	0.0010	0.0010	0.00010	0.000010	0.0025
W7	5/14/2010	9.2	73	71	15	130	0.034	0.0025	0.010	8.10	7.26	0.096	0.00040	0.000010	0.00050	0.0031	0.34	0.00010	0.00050	0.0020	0.00010	0.000010	0.0025
W7	5/16/2010	8.8	86	76	12	120	0.068	0.0025	0.030	8.30	7.32	0.012	0.00030	0.0000050	0.00050	0.0013	0.080	0.00010	0.0010	0.00050	0.00010	0.000010	0.0025
W7	5/17/2010	7.2	80	76	11	130	0.037	0.0025	0.010	8.20	7.52	0.018	0.00040	0.0000050	0.00050	0.0012	0.068	0.00010	0.0010	0.00050	0.00010	0.000010	0.0025
W7	5/19/2010	7.5	99	103	4.0	120	0.020	0.0025	0.060	8.20	7.62	0.016	0.00030	0.0000050	0.00050	0.0017	0.069	0.00010	0.0010	0.00050	0.00030	0.000010	0.0025
W7	5/21/2010	8.7	92	96	22	120	0.017	0.0025	0.050	8.10	8.02	0.013	0.00020	0.000050		0.0010	0.057	0.00010	0.0010	0.0010	0.00040		0.0050
W7	5/23/2010	8.2	98	103	33	150	0.013	0.0060	0.060	8.10	7.61	0.017	0.00020	0.000050		0.0010	0.069	0.00010	0.0010	0.0010	0.00040		0.0050
W7	5/25/2010	8.1	125	124	140	140	0.070	0.025	0.080	7.90	7.24	0.024	0.00050	0.000050		0.0030	0.062	0.00010	0.0020	0.0015	0.00050		0.0050
W7	5/27/2010	8.1	113	148	350	140	0.052	0.0025	0.13	8.00	7.53	0.028	0.00050	0.000050		0.0030	0.069	0.00010	0.0020	0.0015	0.00050		0.0050
W7	5/29/2010	6.3	111	129	200	160	0.10	0.011	0.22	8.20	7.73	0.017	0.00050	0.000050		0.0020	0.051	0.00010	0.0020	0.0010	0.00040		0.0050
W7	5/31/2010	7.2	109	130	290	160	0.025	0.023	0.12	8.20	7.83	0.020	0.00020	0.000050		0.0020	0.050	0.00010	0.0020	0.00050	0.00040		0.0050
W7	6/2/2010	4.9	113	135	81	140	0.0050	0.023	0.14	8.20	7.26	0.012	0.00040	0.000050		0.0010	0.030	0.00010	0.0020	0.00050	0.00040		0.0050
W7	6/4/2010	6.4	111	125	60	140	0.34	0.014	0.16	8.20	7.48	0.013	0.00040	0.000050		0.0010	0.030	0.00010	0.0020	0.00050	0.00040		0.0050
W7	6/6/2010		128	133	39	170	0.10	0.013	0.16	8.20		0.0050	0.00020	0.000050		0.0010	0.024	0.00010	0.0020	0.00050	0.00040		0.0050
W7	6/14/2010	5.9	122	128	43	150	0.016	0.0025	0.16	8.29		0.011	0.00020	0.000050		0.00050	0.025	0.00010	0.0020	0.00050	0.00040		0.0050
W7	6/16/2010	6.1	120	128	17	150	0.016	0.0025	0.17	8.30	6.78	0.0050	0.00020	0.000050		0.00050	0.026	0.00010	0.0020	0.00050	0.00040		0.0050
W7	6/18/2010	6.2	122	123	28	140		0.0025	0.15	8.17	7.03	0.0050	0.00020	0.000050		0.00050	0.023	0.00010	0.0020	0.00050	0.00040		0.0050
W7	6/20/2010	7.6	112	118	33	140		0.0080	0.13	8.20	7.10	0.014	0.00020	0.000050		0.0010	0.028	0.00010	0.0020	0.00050	0.00040		0.0050
W7	6/22/2010	10	98	115	300	140	0.014	0.0025	0.12	8.03	7.92	0.033	0.00050	0.000050		0.0020	0.082	0.00010	0.0010	0.0010	0.00040		0.0050
W7	6/24/2010	10	100	116	64	130	0.012	0.0070	0.070	8.03	7.95	0.018	0.00020	0.000050		0.0010	0.056	0.00010	0.0010	0.0010	0.00040		0.0050
W7	6/26/2010	9.3	106	106	21	120	0.0090	0.0025	0.10	8.26	8.03	0.012	0.00040	0.000050		0.0010	0.038	0.00010	0.0010	0.0040	0.00040		0.0050
W7	6/28/2010	8.7	114	107	9.0	150	0.022	0.0025	0.13	8.03	7.89	0.0050	0.00020	0.000050		0.0010	0.028	0.00010	0.0010	0.00050	0.00040		0.0050
W7	7/5/2010	16	93	106	53	150	0.027	0.0025	0.070	8.06	8.02	0.019	0.00020	0.000050		0.0020	0.10	0.00010	0.0010	0.0010	0.00040		0.0050
W7	8/5/2010	13	118	116		170					7.91	0.011	0.00040	0.000050		0.0020	0.11	0.00010	0.0010	0.0010	0.00040		0.0050
W7	9/12/2010	17	107	104	11	210	0.039	0.0025	0.030	7.98	8.02	0.016	0.00050	0.000050		0.0010	0.23	0.00010	0.0010	0.0020	0.00040		0.0050
W7	10/8/2010	13	111	126	3.0	150	0.069	0.0025	0.070	8.10	7.85	0.011	0.00040	0.0000050	0.00050	0.0012	0.13	0.00010	0.0010	0.0010	0.00010	0.000010	0.0025
W7	4/27/2011	23	43	43	0.5	72	0.010	0.0025	0.010	7.41		0.057	0.00030	0.000030	0.00050	0.0077	0.15	0.00010	0.00050	0.0010	0.000050	0.000010	0.0050
W7	5/2/2011	22	32	33	7.0	40		0.0025	0.010	7.11		0.076	0.00020	0.000040	0.00050	0.0070	0.17	0.00010	0.00050	0.0010	0.000050	0.000010	0.010
W7	6/1/2011	11	92	94	23	120	0.012			8.09		0.013	0.00030	0.0000050	0.00050	0.0016	0.099	0.00010	0.0010	0.0010	0.00010	0.000010	0.0025
W7	6/16/2011	13	97	108	85	130	0.013	0.0025	0.10	8.00		0.022	0.00040	0.0000050	0.00050	0.0019	0.21	0.00010	0.0010	0.0020	0.00010	0.000010	0.0025
W7	7/1/2011	21	77	88	150	120	0.016	0.0070	0.050	7.76		0.034	0.00070	0.000010	0.00050	0.0031	0.44	0.00010	0.00050	0.0020	0.00010	0.000010	0.0025
W7	8/2/2011	24	59	86	400	110		0.0025	0.060	7.67		0.051	0.00070	0.000020	0.00050	0.0041	0.61	0.00010	0.00050	0.0030	0.00010	0.000010	0.0025
W7	8/16/2011	18	98	97	22	150	0.015	0.0025	0.060	7.90		0.036	0.00070	0.0000050	0.00050	0.0019	0.59	0.00010	0.00050	0.0020	0.00010	0.000010	0.0025
W7	8/19/2011	19	93	93	22	140	0.0060	0.0025	0.040	8.12		0.033	0.00080	0.000010	0.00050	0.0020	0.56	0.00010	0.00050	0.0020	0.00010	0.000010	0.0025
W7	9/12/2011	13	118	119	1.0	170	0.025	0.0025	0.050	8.03	7.92	0.024	0.00070	0.000010	0.00050	0.0016	0.62	0.00010	0.00050	0.0020	0.00010	0.000010	0.0025

Table D.2: Background data for Option 1 with median and 95th Percentile values.

Station	Date	DOC (mg/L)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	TSS (mg/L)	TDS (mg/L)	Ammonia (mg/L)	Nitrite (mg/L)	Nitrate (mg/L)	Lab pH (pH Units)	Field pH (pH Units)	Dissolved Aluminum (mg/L)	Dissolved Arsenic (mg/L)	Dissolved Cadmium (mg/L)	Dissolved Chromium (mg/L)	Dissolved Copper (mg/L)	Dissolved Iron (mg/L)	Dissolved Lead (mg/L)	Dissolved Molybdenum (mg/L)	Dissolved Nickel (mg/L)	Dissolved Selenium (mg/L)	Dissolved Silver (mg/L)	Dissolved Zinc (mg/L)
W7	10/6/2011	12	110	113	2.0	150	0.019	0.0025	0.11	8.07	7.65	0.013	0.00060	0.000030	0.00050	0.0012	0.35	0.00010	0.0010	0.0010	0.00010	0.000010	0.0025
W7	11/22/2011	7.7	148	144	27	156	0.021	0.0025	0.14	7.96		0.0076	0.00040	0.0000050	0.00050	0.0011	0.17	0.00010	0.0013	0.0012	0.00016	0.000010	0.0025
W7	1/24/2012	5.3	169	158	2.0	226	0.0050	0.0025	0.32	8.13		0.0053	0.00041	0.000035	0.00050	0.0011	0.094	0.00010	0.0015	0.00050	0.00043	0.000010	0.0025
W7	2/25/2012	6.1	165	160	2.0	216	0.0060	0.0025	0.29	7.99	7.16	0.021	0.00033	0.000064	0.00050	0.015	0.049	0.00024	0.0014	0.00050	0.00034	0.000010	0.0083
W7	3/5/2012	5.2	161	159	3.2	176	0.0067	0.0025	0.20	8.17	6.72	0.0031	0.00039	0.0000050	0.00050	0.0012	0.012	0.00010	0.0011	0.00050	0.00031	0.000010	0.0025
W7	4/24/2012	14	39	37	11	100	0.0025	0.010	0.010	7.63	7.43	0.050	0.00048	0.000023	0.00050	0.0052	0.48	0.00010	0.00050	0.0012	0.000050	0.000010	0.0025
W7	5/6/2012	17	46	45	3.3	84	0.011	0.0025	0.010	5.90	7.10	0.037	0.00044	0.000015	0.00050	0.0029	0.44	0.00010	0.00050	0.0012	0.000050	0.000010	0.0025
W7	6/19/2012	16	92	105	165	128	0.18	0.015	0.083	8.12	7.59	0.027	0.00070	0.0000050	0.00050	0.0016	0.56	0.00010	0.0013	0.0020	0.00015	0.000010	0.0025
W7	7/17/2012	13	128	124	34	164	0.065	0.0025	0.089	8.15	7.76	0.020	0.00086	0.000012	0.00050	0.0013	0.97	0.00010	0.0015	0.0016	0.00013	0.000010	0.0025
W7	7/30/2012	13	135	128	9.2	180	0.13	0.0025	0.12	8.20	7.86	0.017	0.00095	0.0000050	0.00050	0.0012	0.96	0.00010	0.0015	0.0015	0.00016	0.000010	0.0025
W7	8/8/2012	13	141	141	3.3	192	0.014	0.0025	0.13	8.26	8.44	0.018	0.0010	0.0000050	0.00050	0.0011	1.1	0.00010	0.0016	0.0017	0.00014	0.000010	0.0025
W7	8/23/2012	10	142	151	1.5	200	0.049	0.025	0.21	8.25	7.79	0.018	0.00088	0.0000050	0.00050	0.0017	0.98	0.00010	0.0016	0.0016	0.00015	0.000010	0.0025
W7	9/13/2012	16	116	121	15	154	0.0087	0.0025	0.078	8.13	7.98	0.024	0.00074	0.0000050	0.00050	0.0024	0.73	0.00010	0.00050	0.0036	0.00014	0.000010	0.0025
W7	10/16/2012	13	129	143	10	166	0.028	0.0025	0.13	8.17	7.89	0.011	0.00060	0.0000050	0.00050	0.00094	0.60	0.00010	0.0011	0.0015	0.00014	0.000010	0.0025
W7	11/2/2012	8.2	158	153	2.2	214	0.026	0.0025	0.22	7.93		0.0095	0.00053	0.000031	0.00050	0.0024	0.18	0.00021	0.0013	0.0012	0.00018	0.000010	0.0025
W7	12/29/2012	10	223	206	0.50	262	0.025	0.0025	0.063	8.13	7.97	0.0061	0.00051	0.0000050	0.00050	0.0023	0.030	0.00010	0.0012	0.0011	0.00045	0.000010	0.0025
W7	4/30/2013	4.6	192	198	5.8	246	0.0082	0.0025	0.13	8.27	8.18	0.0031	0.00025	0.0000050	0.00050	0.0011	0.014	0.00010	0.0013	0.00050	0.00043	0.000010	0.0025
W7	5/14/2013	15	33	36	55	70	0.066	0.0025	0.010	7.58	8.20	0.044	0.00029	0.000010	0.00050	0.0037	0.31	0.00010	0.00050	0.0010	0.000050	0.000010	0.0025
W7	6/19/2013	11	114	115	11	158	0.046	0.0025	0.14	8.08	8.60	0.013	0.00052	0.0000050	0.00050	0.0012	0.29	0.00010	0.0016	0.0013	0.00015	0.000010	0.0025
W7	7/13/2013	10	132	137	5.0	156	0.023	0.0025	0.17	8.10	7.74	0.014	0.00053	0.0000050	0.00050	0.0012	0.42	0.00010	0.0015	0.0013	0.00015	0.000010	0.0025
W7	8/21/2013	10	140	144	3.8	172	0.051	0.0025	0.16	8.23	7.81	0.012	0.00056	0.0000050	0.00050	0.0010	0.30	0.00010	0.0015	0.0012	0.00015	0.000010	0.0025
W7	9/6/2013	11	138	154	12	280	0.017	0.0025	0.13	8.02	7.75	0.016	0.00081	0.0000050	0.00050	0.0011	0.66	0.00010	0.0013	0.0016	0.00017	0.000010	0.0025
W7	10/21/2013	12	147	126	7.4	168	0.048	0.0025	0.17	8.16	7.58	0.0093	0.00067	0.0000050	0.00050	0.0011	0.55	0.00010	0.0012	0.0012	0.00017	0.000010	0.0025
W7	11/16/2013	7.5	153	164	1.5	164	0.027	0.0025	0.26	8.24	7.69	0.0053	0.00048	0.0000050	0.00050	0.00080	0.17	0.00010	0.0016	0.00050	0.00024	0.000010	0.0025
W7	12/7/2013	7.3	168	189	0.50	220	0.025	0.0025	0.21	8.14	6.30	0.0035	0.00053	0.0000050	0.00050	0.00078	0.11	0.00010	0.0011	0.00050	0.00024	0.000010	0.0025
C4	5/6/2012	11	44	119	1,260	84	0.088	0.0025	0.029	7.59	7.51	0.062	0.00104	0.000029	0.00050	0.0031	1.1	0.00010	0.00050	0.0023	0.000050	0.000010	0.0025
C4	8/1/2012	16	117	133	246	166	0.089	0.018	0.074	8.04	7.75	0.059	0.00184	0.0000050	0.00050	0.0018	1.5	0.00010	0.0011	0.0031	0.00011	0.000010	0.0025
C4	8/8/2012		125	142	253	170	0.090	0.012	0.095	8.14	7.81	0.043	0.00203	0.0000050	0.00050	0.0013	1.6	0.00010	0.0011	0.0027	0.00011	0.000010	0.0025
C4	8/25/2012	0.25	130	173	855	192	0.0050	0.0025	0.010	7.94	7.81	0.040	0.0020	0.0000050	0.00050	0.0018	1.7	0.00010	0.0012	0.0029	0.000050	0.000010	0.0025
C4	6/22/2013	17	126	222	886	204	0.36	0.0066	0.37	7.97	7.68	0.034	0.0023	0.0000050	0.00050	0.0017	1.2	0.00010	0.0018	0.0027	0.00013	0.000010	0.0025
C4	7/19/2013		128	200	487	184	0.26	0.0069	0.082	8.12	8.20	0.032	0.0021	0.0000050	0.00050	0.0014	1.6	0.00010	0.0017	0.0025	0.00012	0.000010	0.0025
C4	8/22/2013	19	123	199	784	182	0.20	0.0084	0.086	7.98	7.88	0.046	0.0018	0.0000050	0.00050	0.0014	1.4	0.00010	0.0013	0.0027	0.00012	0.000010	0.0025
C4	9/6/2013	15	125	166	172	176	0.12	0.0025	0.070	7.93	7.69	0.041	0.0020	0.0000050	0.00050	0.0016	1.8	0.00010	0.0010	0.0028	0.000050	0.000010	0.0025
C4	10/18/2013	13	129	141	16	217	0.086	0.0025	0.067	8.21	7.16	0.024	0.0011	0.0000050	0.00050	0.0012	1.2	0.00010	0.00085	0.0024	0.000050	0.000010	0.0025
C4	11/17/2013	32	258	264	22	272	0.28	0.025	0.10	7.90	7.17	0.033	0.0056	0.000022	0.0013	0.00095	14	0.00010	0.0011	0.0061	0.00024	0.000010	0.0025
C10	5/12/2012	13	67	215	2,210	132	0.18	0.011	0.047	7.73	7.89	0.058	0.0012	0.000018	0.00050	0.0021	1.6	0.00025	0.00050	0.0019	0.00010	0.000010	0.0025
C10	7/30/2012	18	130	207	995	180	0.20	0.023	0.18	8.23		0.057	0.0014	0.0000050	0.00050	0.0020	1.2	0.00067	0.0014	0.0023	0.00014	0.000010	0.0025
C10	8/10/2012		135	245	1,320	216	0.12	0.011	0.18	8.11	8.71	0.051	0.0013	0.0000050	0.00050	0.0019	0.88	0.00010	0.0019	0.0022	0.00010	0.000010	0.0025
C10	8/27/2012	18	141	264	1,650	184	0.0025	0.0086	0.095	8.37	8.30	0.026	0.0015	0.0000050	0.00050	0.0015	1.0	0.00010	0.0011	0.0023	0.00012	0.000010	0.0025
C10	6/22/2013	15	148	151	43	192	0.19	0.020	0.15	8.26	8.29	0.079	0.0012	0.000019	0.00050	0.0031	0.80	0.00010	0.0012	0.0025	0.00011	0.000010	0.010
C10	7/23/2013		125	156	199	174	0.14	0.025	0.10	8.19	8.31	0.025	0.0012	0.0000050	0.00050	0.0014	0.54	0.00010	0.0010	0.0022	0.000050	0.000010	0.0025
C10	8/22/2013	17	160	168	66	218	0.056	0.0072	0.14	8.23	8.18	0.020	0.0014	0.0000050	0.00050	0.0013	0.99	0.00010	0.0015	0.0023	0.000050	0.000010	0.0025
C10	9/8/2013	16	150	163	24	196	0.051	0.0073	0.18	8.04	8.16	0.025	0.0016	0.0000075	0.0011	0.0013	1.3	0.00010	0.00050	0.0025	0.000050	0.000010	0.0025
C10	10/18/2013	14	136	160	61	212	0.10	0.0025	0.17	8.25	7.78	0.011	0.00089	0.0000050	0.00050	0.00091	0.76	0.00010	0.00050	0.0019	0.000050	0.000010	0.0025
Median:		**11**	**117**	**121**	**11**	**150**	**0.021**	**0.0025**	**0.067**	**8.00**	**7.76**	**0.016**	**0.00044**	**0.0000085**	**0.00050**	**0.0016**	**0.11**	**0.00010**	**0.0010**	**0.0011**	**0.00020**	**0.000010**	**0.0025**
95th Percentile:		**22**	**162**	**200**	**772**	**217**	**0.15**	**0.025**	**0.21**	**8.26**	**8.30**	**0.071**	**0.0015**	**0.000050**	**0.0011**	**0.0050**	**1.2**	**0.00025**	**0.0020**	**0.0027**	**0.00050**	**0.000010**	**0.0060**

Table D.3: Background data for Option 2 with median and 95th Percentile values.

Year	Month	Day	DOC (mg/L)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	TSS (mg/L)	TDS (mg/L)	Ammonia (mg/L)	Nitrite (mg/L)	Nitrate (mg/L)	Lab pH (pH Units)	Field pH (pH Units)	Dissolved Aluminum (mg/L)	Dissolved Arsenic (mg/L)	Dissolved Cadmium (mg/L)	Dissolved Chromium (mg/L)	Dissolved Copper (mg/L)	Dissolved Iron (mg/L)	Dissolved Lead (mg/L)	Dissolved Molybdenum (mg/L)	Dissolved Nickel (mg/L)	Dissolved Selenium (mg/L)	Dissolved Silver (mg/L)	Dissolved Zinc (mg/L)
2005	May	27			78	22	118	0.016	0.0012	0.038	7.92		0.028	0.00029	0.000025	0.00050	0.0034	0.15	0.00025	0.00050	0.0015	0.00050	0.000010	0.0025
	June	30				5.4	160	0.015	0.00067	0.052	7.83		0.026	0.00046	0.000025	0.00030	0.0023	0.16	0.000064	0.00085	0.0011	0.00050	0.0000050	0.0091
	July	28/29			138	11	173	0.010	0.00050	0.042	7.73		0.020	0.00041	0.000025	0.00025	0.0033	0.16	0.000075	0.0011	0.0011	0.00050	0.0000050	0.0030
	August	29/30			110	4.9	157	0.029	0.00066	0.010	8.10		0.022	0.00043	0.000010	0.00050	0.0026	0.27	0.00025	0.00083	0.0014	0.00050	0.000010	0.0029
	September	28			111	58	151	0.012	0.00057	0.025	7.83		0.024	0.00039	0.0000085	0.00050	0.0027	0.31	0.00025	0.00059	0.0013	0.00050	0.000010	0.0025
	October	15			111	5.5	151	0.015	0.00050	0.060	7.64		0.033	0.00044	0.000025	0.00025	0.0029	0.25	0.000025	0.00069	0.0012	0.00050	0.0000050	0.0013
2006		2				410	100	0.042	0.0010	0.035	7.63		0.048	0.00053	0.000025	0.00040	0.0019	0.32	0.000040	0.00050	0.0020	0.00050	0.0000050	0.0018
		8																						
	June	15		100		53	114	0.027		0.18	7.86		0.013	0.00040	0.0000050	0.00025	0.0010	0.090	0.000050	0.0010	0.0010	0.00020		0.00050
		23		122		328	176	0.028		0.23	7.78		0.011	0.00050	0.0000050	0.00070	0.00050	0.090	0.000050	0.0010	0.0013	0.00030		0.0020
		28		107		103	120	0.010		0.11	7.90		0.41	0.00080	0.000030	0.0015	0.0030	0.060	0.00050	0.00050	0.0036	0.00010		0.0050
	July	7		110		3.0	184			0.21	7.91		0.0025	0.00060	0.0000050	0.0011	0.0010	0.060	0.000050	0.0020	0.0015	0.00010		0.0010
		12		110		1.0		0.014		0.18	7.77		0.0060	0.00050	0.0000050	0.0010	0.0010	0.060	0.000050	0.0010	0.00070	0.00020		0.00050
		20		130		2.0		0.012		0.19	7.88		0.0060	0.00040	0.0000050	0.00070	0.00050	0.11	0.00020	0.0010	0.00060	0.00010		0.00050
		26		128		9.0		0.016		0.18	7.87		0.0080	0.00040	0.0000050	0.00025	0.0020	0.10	0.000050	0.00050	0.0021	0.00010		0.00050
	August	2		128		4.0		0.015		0.15	7.79		0.0080	0.00010	0.0000050	0.00025	0.0010	0.070	0.000050	0.00050	0.00060	0.00010		0.00050
		10		126		1.0		0.014		0.16	7.80		0.0090	0.00050	0.0000050	0.00070	0.0010	0.080	0.000050	0.0010	0.00060	0.00010		0.0010
		25		125			170	0.027		0.068	7.97		0.012	0.00040	0.0000050	0.00060	0.0010	0.080	0.000050	0.0010	0.00090	0.00010		0.0030
		30		132		1.0	200	0.012		0.11	7.79		0.013	0.00040	0.0000050	0.00080	0.0020	0.10	0.000050	0.0010	0.0012	0.00010		0.0050
	September	6		120		4.0		0.011		0.048	7.87		0.0065	0.00050	0.0000050	0.00070	0.0015	0.11	0.000050	0.00075	0.00080	0.00010		0.0020
		13		130		5.0		0.011		0.094	7.84		0.0080	0.00030	0.0000050	0.00090	0.0010	0.11	0.000050	0.0010	0.00090	0.00010		0.0010
		20		132		3.0		0.008		0.11	7.85		0.0070	0.00050	0.00028	0.00070	0.0010	0.090	0.000050	0.0010	0.00025	0.00020		0.00050
		28		129			182	0.011		0.080	7.26		0.011	0.00040	0.0000050	0.00080	0.0010	0.080	0.000050	0.0010	0.00080	0.00010		0.0020
	October	4		126		2.0		0.079		0.10	7.94		0.0025	0.00030	0.0000050	0.00025	0.00050	0.11	0.000050	0.0010	0.00070	0.00010		0.0010
		12		132		4.0	164	0.024		0.12	7.39		0.0070	0.00040	0.0000050	0.00025	0.0010	0.090	0.000050	0.0010	0.00060	0.00030		0.0020
2007	June	5			113	75	162	0.025	0.033	0.050	7.97													
		20	8.2	110	81	11	147	0.025	0.025	0.20	7.81		0.014	0.00040	0.0000050	0.00060	0.0020	0.050	0.000050	0.0010	0.00025	0.00010		0.0040
	July	18	14	128	125	5.7	179	0.025	0.025	0.30	7.78		0.011	0.00050	0.0000050	0.0010	0.0020	0.040	0.000050	0.0010	0.00070	0.00010		0.0060
	August	24			124	4.7	192	0.025	0.025	0.20	8.04													
	September	18	11	110	93	14	177	0.025	0.010	0.030	7.98		0.0070	0.00040	0.0000050	0.0016	0.0010	0.080	0.000050	0.0020	0.00080	0.00010		0.0020
	October	30			129	6.5	174	0.025	0.14	0.010	7.73													
2008	April	22		187		1.0	268	0.025	0.055	0.010	8.40		0.010	0.00027	0.000040	0.00097	0.0012	0.030	0.000067	0.0017	0.0010	0.00030		0.0017
	June	3		92	98	15	137	0.025	0.040	0.037	8.02		0.0050	0.00010	0.000040	0.0013	0.00050	0.050	0.000050	0.0012	0.0010	0.00030	0.0000050	0.0020
		17			86	14	138	0.025	0.060	0.075	7.89		0.020	0.00010	0.000040	0.0013	0.0010	0.070	0.000050	0.0015	0.0010	0.0022	0.0000050	0.0020
	August	6	9.0	123	124	105	167	0.025	0.0050	0.14	7.92		0.014	0.00040	0.0000050	0.00060	0.0020	0.070	0.000050	0.0010	0.0010	0.00010	0.000060	0.0060
	September	18	18			16	160	0.025	0.0050	0.018	7.85		0.041	0.00035	0.0000050	0.0015	0.0030	0.15	0.00088	0.00050	0.0015	0.00075	0.0000050	0.0030
	October	28	11	123		5.7	183	0.025			7.84		0.0090	0.00050	0.0000050	0.00050	0.0010	0.078	0.000050	0.0011	0.0010	0.00030	0.0000050	0.0010
2009	April	30	18	20		1.0	52				6.84	7.60	0.053	0.00010	0.000080	0.00060	0.0050	0.070	0.00010	0.00013	0.00050	0.00030	0.0000050	0.0090
	June	8	9.2	93		32	156				7.90	7.74	0.013	0.00030	0.0000050	0.00020	0.0020	0.050	0.000050	0.0013	0.0010	0.00030	0.0000050	0.0020
	July	27		137		4.0	184				8.07													
	August	6		121		2.0	202			0.18	8.35	7.50												
		25	8.2									8.04												
		31		110									0.010	0.00040	0.0000050	0.0012	0.0040	0.10	0.00040	0.0011	0.0010	0.00030	0.0000050	0.0010
	October	7	7.8	125	141	3.7	140	0.010	0.0025	0.14	7.99	8.24	0.0070	0.00030	0.000040	0.00050	0.0012	0.046	0.00010	0.0010	0.00050	0.00020	0.000010	0.0025
		25	8.7	126	141	2.0	120	0.010	0.0025	0.15	8.05	8.23	0.0070	0.00030	0.000040	0.00050	0.0012	0.046	0.00010	0.0010	0.00050	0.00020	0.000010	0.0025

Table D.3: Background data for Option 2 with median and 95th Percentile values.

Year	Month	Day	DOC (mg/L)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	TSS (mg/L)	TDS (mg/L)	Ammonia (mg/L)	Nitrite (mg/L)	Nitrate (mg/L)	Lab pH (pH Units)	Field pH (pH Units)	Dissolved Aluminum (mg/L)	Dissolved Arsenic (mg/L)	Dissolved Cadmium (mg/L)	Dissolved Chromium (mg/L)	Dissolved Copper (mg/L)	Dissolved Iron (mg/L)	Dissolved Lead (mg/L)	Dissolved Molybdenum (mg/L)	Dissolved Nickel (mg/L)	Dissolved Selenium (mg/L)	Dissolved Silver (mg/L)	Dissolved Zinc (mg/L)
2010	April	19			80	3.5	142				7.81	7.31	0.019	0.00030	0.000020	0.00040	0.0060	0.090	0.00030	0.00080	0.00050	0.00030	0.0000050	0.0050
		27/29	21	36	36	2.0	62	0.016	0.0038	0.015	7.55	7.69	0.070	0.00030	0.0000050	0.00050	0.0038	0.17	0.00010	0.00050	0.0010	0.000050	0.000010	0.0025
	May	1-7	17	48	50	14	73	0.026	0.0050	0.048	7.75	7.61	0.042	0.00043	0.000010	0.00050	0.0028	0.39	0.00010	0.00050	0.0013	0.000050	0.000010	0.0025
		9-14	11	79	79	5.3	123	0.019	0.0034	0.060	8.05	7.78	0.039	0.00040	0.000010	0.00050	0.0022	0.27	0.00010	0.00063	0.0013	0.00010	0.000010	0.0025
		16-21	8.1	89	88	12	123	0.036	0.0025	0.038	8.20	7.62	0.015	0.00030	0.000016	0.00050	0.0013	0.069	0.00010	0.0010	0.00063	0.00023	0.000010	0.0031
		23-29	7.7	112	126	181	148	0.059	0.011	0.12	8.05	7.53	0.022	0.00043	0.000050		0.0023	0.063	0.00010	0.0018	0.0013	0.00045		0.0050
	May/June	31-6	6.2	115	131	118	153	0.12	0.018	0.15	8.20	7.52	0.013	0.00030	0.000050		0.0013	0.034	0.00010	0.0020	0.00050	0.00040		0.0050
	June	14-20	6.5	119	124	30	145	0.016	0.0039	0.15	8.24	6.97	0.0088	0.00020	0.000050		0.00063	0.026	0.00010	0.0020	0.00050	0.00040		0.0050
		22-28	10	104	111	99	135	0.014	0.0036	0.11	8.09	7.95	0.017	0.00033	0.000050		0.0013	0.051	0.00010	0.0010	0.0016	0.00040		0.0050
	July	5	16	93	106	53	150	0.027	0.0025	0.070	8.06	8.02	0.019	0.00020	0.000050		0.0020	0.10	0.00010	0.0010	0.0010	0.00040		0.0050
	August	5	13	118	116		170					7.91	0.011	0.00040	0.000050		0.0020	0.11	0.00010	0.0010	0.0010	0.00040		0.0050
	September	12	17	107	104	11	210	0.039	0.0025	0.030	7.98	8.02	0.016	0.00050	0.000050		0.0010	0.23	0.00010	0.0010	0.0020	0.00040		0.0050
	October	8	13	111	126	3.0	150	0.069	0.0025	0.070	8.10	7.85	0.011	0.00040	0.0000050	0.00050	0.0012	0.13	0.00010	0.0010	0.0010	0.00010	0.000010	0.0025
2011	April	27	23	43	43	0.50	72	0.010	0.0025	0.010	7.41		0.057	0.00030	0.000030	0.00050	0.0077	0.15	0.00010	0.00050	0.0010	0.000050	0.000010	0.0050
	May	2	22	32	33	7.0	40		0.0025	0.010	7.11		0.076	0.00020	0.000040	0.00050	0.0070	0.17	0.00010	0.00050	0.0010	0.000050	0.000010	0.010
	June	1	11	92	94	23	120	0.012			8.09		0.013	0.00030	0.0000050	0.00050	0.0016	0.10	0.00010	0.0010	0.0010	0.00010	0.000010	0.0025
		16	13	97	108	85	130	0.013	0.0025	0.10	8.00		0.022	0.00040	0.0000050	0.00050	0.0019	0.21	0.00010	0.0010	0.0020	0.00010	0.000010	0.0025
	July	1	21	77	88	150	120	0.016	0.0070	0.050	7.76		0.034	0.00070	0.000010	0.00050	0.0031	0.44	0.00010	0.00050	0.0020	0.00010	0.000010	0.0025
	August	2	24	59	86	400	110		0.0025	0.060	7.67		0.051	0.00070	0.000020	0.00050	0.0041	0.61	0.00010	0.00050	0.0030	0.00010	0.000010	0.0025
		16/19	19	95	95	22	145	0.011	0.0025	0.050	8.01		0.035	0.00075	0.0000075	0.00050	0.0020	0.57	0.00010	0.00050	0.0020	0.00010	0.000010	0.0025
	September	12	13	118	119	1.0	170	0.025	0.0025	0.050	8.03	7.92	0.024	0.00070	0.000010	0.00050	0.0016	0.62	0.00010	0.00050	0.0020	0.00010	0.000010	0.0025
	October	6	12	110	113	2.0	150	0.019	0.0025	0.11	8.07	7.65	0.013	0.00060	0.000030	0.00050	0.0012	0.35	0.00010	0.0010	0.0010	0.00010	0.000010	0.0025
	November	22	7.7	148	144	27	156	0.021	0.0025	0.14	7.96		0.0076	0.00040	0.0000050	0.00050	0.0011	0.17	0.00010	0.0013	0.0012	0.00016	0.000010	0.0025
2012	January	24	5.3	169	158	2.0	226	0.0050	0.0025	0.32	8.13		0.0053	0.00041	0.000035	0.00050	0.0011	0.094	0.00010	0.0015	0.00050	0.00043	0.000010	0.0025
	February	25	6.1	165	160	2.0	216	0.0060	0.0025	0.29	7.99	7.16	0.021	0.00033	0.000064	0.00050	0.015	0.049	0.00024	0.0014	0.00050	0.00034	0.000010	0.0083
	March	5	5.2	161	159	3.2	176	0.0067	0.0025	0.20	8.17	6.72	0.0031	0.00039	0.0000050	0.00050	0.0012	0.012	0.00010	0.0011	0.00050	0.00031	0.000010	0.0025
	April	24	14	39	37	11	100	0.0025	0.010	0.010	7.63	7.43	0.050	0.00048	0.000023	0.00050	0.0052	0.48	0.00010	0.00050	0.0012	0.000050	0.000010	0.0025
	May	6/12	14	52	126	1158	100	0.093	0.0055	0.029	7.07	7.50	0.053	0.00089	0.000021	0.00050	0.0027	1.0	0.00015	0.00050	0.0018	0.000067	0.000010	0.0025
	June	19	16	92	105	165	128	0.18	0.015	0.083	8.12	7.59	0.027	0.00070	0.0000050	0.00050	0.0016	0.56	0.00010	0.0013	0.00200	0.00015	0.000010	0.0025
	July	17	11	116	144	582	143	0.10	0.0025	0.058	8.07	7.84	0.018	0.00070	0.0000085	0.00050	0.0012	0.52	0.00010	0.0010	0.0013	0.00013	0.000010	0.0025
		30	13	122	160	539	161	0.13	0.011	0.12	8.17	7.88	0.031	0.00099	0.0000050	0.00050	0.0014	0.73	0.00029	0.0011	0.0014	0.00012	0.000010	0.0025
	August	1	16	117	133	246	166	0.089	0.018	0.074	8.04	7.75	0.059	0.0018	0.0000050	0.00050	0.0018	1.5	0.00010	0.0011	0.0031	0.00011	0.000010	0.0025
		8-10	13	128	166	536	181	0.062	0.0081	0.11	8.17	8.26	0.032	0.0013	0.0000050	0.00050	0.0013	0.91	0.00010	0.0013	0.0019	0.00011	0.000010	0.0025
		23-27	9.0	129	187	917	179	0.020	0.010	0.090	8.14	7.92	0.024	0.0013	0.0000050	0.00050	0.0015	0.93	0.00010	0.0011	0.0020	0.000093	0.000010	0.0025
	September	13	15	102	107	33	136	0.010	0.0025	0.056	8.09	7.97	0.020	0.00061	0.0000050	0.00050	0.0019	0.42	0.00010	0.00050	0.0025	0.000095	0.000010	0.0025
	October	13/16	11	129	130	27	158	0.022	0.0025	0.084	8.05	8.10	0.011	0.00060	0.0000050	0.00050	0.00094	0.60	0.00010	0.0011	0.0015	0.00014	0.000010	0.0025
	November	2	8.2	158	153	2.2	214	0.026	0.0025	0.22	7.93		0.0095	0.00053	0.000031	0.00050	0.0024	0.18	0.00021	0.0013	0.0012	0.00018	0.000010	0.0025
		12	8.7	153	155	2.2	162	0.025	0.0025	0.010	7.93	6.69	0.0062	0.00039	0.0000050	0.00050	0.0014	0.024	0.00010	0.00050	0.0014	0.00010	0.000010	0.0025
	December	29	10	223	206	0.50	262	0.025	0.0025	0.063	8.13	7.97	0.0061	0.00051	0.0000050	0.00050	0.0023	0.030	0.00010	0.0012	0.0011	0.00045	0.000010	0.0025
2013	April	30	4.6	192	198	5.8	246	0.0082	0.0025	0.13	8.27	8.18	0.0031	0.00025	0.0000050	0.00050	0.0011	0.014	0.00010	0.0013	0.00050	0.00043	0.000010	0.0025
	May	14	19	32	35	34	77	0.10	0.0025	0.010	7.53	7.90	0.051	0.00033	0.000011	0.00050	0.0036	0.245	0.00010	0.00050	0.0012	0.000050	0.000010	0.0025
	June	19/22	13	121	151	278	168	0.16	0.011	0.17	8.09	8.05	0.034	0.0011	0.0000085	0.00050	0.0017	0.590	0.00010	0.0013	0.0019	0.00011	0.000010	0.0045
	July	12/13	9.1	110	116	15	146	0.018	0.0025	0.090	8.00	7.72	0.010	0.00050	0.0000050	0.00050	0.0011	0.26	0.00010	0.0010	0.0012	0.00010	0.000010	0.0025
		19/23		127	178	343	179	0.20	0.016	0.091	8.16	8.26	0.028	0.0016	0.0000050	0.00050	0.0014	1.1	0.00010	0.0014	0.0024	0.000085	0.000010	0.0025
	August	21/22	13	131	154	218	179	0.080	0.0052	0.10	8.12	7.92	0.021	0.0011	0.0000050	0.00050	0.0012	0.71	0.00010	0.0012	0.0018	0.000093	0.000010	0.0025
	September	6/8	12	129	149	53	196	0.048	0.0037	0.10	7.98	7.80	0.022	0.0012	0.0000056	0.00065	0.0012	0.96	0.00010	0.00083	0.0020	0.000080	0.000010	0.0025
	October	7	8.2	107	106	74	176	0.044	0.0025	0.042	8.08	7.66	0.013	0.00051	0.0000050	0.00050	0.00097	0.15	0.00010	0.00050	0.0013	0.000050	0.000010	0.0025
		18/21	13	137	142	28	199	0.078	0.0025	0.13	8.21	7.51	0.015	0.00088	0.0000050	0.00050	0.0011	0.82	0.00010	0.00085	0.0018	0.000090	0.000010	0.0025
	November	16/17	16	177	184	19	185	0.11	0.010	0.12	8.08	7.61	0.015	0.0022	0.000011	0.00077	0.00086	4.8	0.00010	0.0011	0.0025	0.00018	0.000010	0.0025
	December	7	8.0	164	186	2.1	207	0.016	0.0025	0.11	8.18	6.90	0.0038	0.00057	0.0000080	0.00050	0.0012	0.06	0.00010	0.00080	0.00080	0.00015	0.000010	0.0025
Median:			**12**	**120**	**121**	**11**	**158**	**0.025**	**0.0025**	**0.090**	**7.97**	**7.77**	**0.014**	**0.00041**	**0.0000053**	**0.00050**	**0.0015**	**0.11**	**0.00010**	**0.0010**	**0.0011**	**0.00011**	**0.000010**	**0.0025**
95th Percentile:			**21**	**169**	**185**	**485**	**216**	**0.12**	**0.038**	**0.22**	**8.20**	**8.24**	**0.056**	**0.0013**	**0.000050**	**0.0013**	**0.0051**	**0.95**	**0.00028**	**0.0017**	**0.0024**	**0.00050**	**0.000010**	**0.0060**

Table D.4: Background data for Option 3 with median and 95th Percentile values.

Year	Month	DOC (mg/L)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	TSS (mg/L)	TDS (mg/L)	Ammonia (mg/L)	Nitrite (mg/L)	Nitrate (mg/L)	Lab pH (pH Units)	Field pH (pH Units)	Dissolved Aluminum (mg/L)
2005	May			74	19	118	0.017	0.0011	0.034	7.85		0.043
	June				5.4	160	0.015	0.00067	0.052	7.83		0.026
	July			138	11	173	0.010	0.00050	0.042	7.73		0.020
	August			110	4.9	157	0.029	0.00066	0.010	8.10		0.022
	September			111	58	151	0.012	0.00057	0.025	7.83		0.024
	October			111	5.5	151	0.015	0.00050	0.060	7.64		0.033
2006	June		110		206	112	0.026	0.0010	0.083	7.71		0.075
	July		120		3.8	184	0.014		0.19	7.86		0.0056
	August		124		2.0	175	0.017		0.075	7.92		0.011
	September		120		4.0	182	0.010		0.053	7.78		0.0079
	October		129		3.0	164	0.052		0.11	7.67		0.0048
2007	June	8.2	110	97	28	147	0.025	0.027	0.14	7.84		0.014
	July	14	128	125	5.7	179	0.025	0.025	0.30	7.78		0.011
	August			124	4.7	192	0.025	0.025	0.20	8.04		
	September	11	110	93	14	177	0.025	0.010	0.030	7.98		0.0070
	October			129	6.5	174	0.025	0.14	0.010	7.73		
2008	April		187		1.0	268	0.025	0.055	0.010	8.40		0.010
	June		92	92	15	133	0.025	0.058	0.049	7.90		0.013
	August	9.0	123	124	105	167	0.025	0.005	0.143	7.92		0.014
	September	18			16	160	0.025	0.005	0.018	7.85		0.041
	October	11	123		5.7	183	0.025			7.84		0.0090
2009	April	18	20		1.0	52				6.84	7.60	0.053
	June	9.2	93		32	156				7.90	7.74	0.013
	July		137		4.0	184				8.07		
	August	12	96		9.8	149			0.18	7.79	7.72	0.025
	October	8.3	125	141	2.8	130	0.010	0.0025	0.14	8.02	8.24	0.007
2010	April	21	36	50	2.5	89	0.016	0.0038	0.015	7.64	7.56	0.053
	May	11	83	88	67	119	0.034	0.0065	0.070	8.02	7.64	0.029
	June	7.6	113	121	63	143	0.059	0.0073	0.14	8.17	7.49	0.012
	July	16	93	106	53	150	0.027	0.0025	0.070	8.06	8.02	0.019
	August	13	118	116		170					7.91	0.011
	September	17	107	104	11	210	0.039	0.0025	0.030	7.98	8.02	0.016
	October	13	111	126	3.0	150	0.069	0.0025	0.070	8.10	7.85	0.011
2011	April	23	43	43	0.50	72	0.010	0.0025	0.010	7.41		0.057
	May	22	32	33	7.0	40		0.0025	0.010	7.11		0.076
	June	17	66	69	29	91	0.012	0.0025	0.040	7.65		0.042
	July	21	77	88	150	120	0.016	0.0070	0.050	7.76		0.034
	August	21	82	91	149	130	0.012	0.0036	0.053	7.86		0.039
	September	13	118	119	1.0	170	0.025	0.0025	0.050	8.03	7.92	0.024
	October	12	110	113	2.0	150	0.019	0.0025	0.11	8.07	7.65	0.013
	November	7.7	148	144	27	156	0.021	0.0025	0.14	7.96		0.0076
2012	January	5.3	169	158	2.0	226	0.0050	0.0025	0.32	8.13		0.0053
	February	6.1	165	160	2.0	216	0.0060	0.0025	0.29	7.99	7.16	0.021
	March	5.2	161	159	3.2	176	0.0067	0.0025	0.20	8.17	6.72	0.0031
	April	14	39	37	11.2	100	0.0025	0.010	0.010	7.63	7.43	0.050
	May	14	52	126	1158	100	0.093	0.0055	0.029	7.07	7.50	0.053
	June	16	92	105	165	128	0.18	0.015	0.083	8.12	7.59	0.027
	July	13	121	162	630	158	0.134	0.010	0.11	8.15	7.86	0.031
	August	11	128	175	700	178	0.045	0.0097	0.10	8.16	8.08	0.030
	September	15	102	107	33	136	0.0099	0.0025	0.056	8.09	7.97	0.020
	October	11	129	130	27	158	0.022	0.0025	0.084	8.05	8.10	0.011
	November	8.5	156	154	2.2	188	0.026	0.0025	0.11	7.93	6.69	0.0079
	December	10	223	206	0.50	262	0.025	0.0025	0.063	8.13	7.97	0.0061
2013	April	4.6	192	198	5.8	246	0.0082	0.0025	0.13	8.27	8.18	0.0031
	May	19	32	35	34	77	0.098	0.0025	0.010	7.53	7.90	0.051
	June	13	121	151	278	168	0.16	0.011	0.17	8.09	8.05	0.034
	July	9.1	118	147	179	163	0.11	0.0092	0.091	8.08	7.99	0.019
	August	13	131	154	218	179	0.080	0.0052	0.097	8.12	7.92	0.021
	September	12	129	149	53	196	0.048	0.0037	0.10	7.98	7.80	0.022
	October	12	130	133	40	193	0.069	0.0025	0.11	8.18	7.55	0.014
	November	16	177	184	19	185	0.11	0.010	0.12	8.08	7.61	0.015
	December	8.0	164	186	2.1	207	0.016	0.0025	0.11	8.18	6.90	0.0038
Median:		**12**	**118**	**124**	**11**	**160**	**0.025**	**0.0025**	**0.075**	**7.96**	**7.80**	**0.019**
95th Percentile:		**21**	**181**	**185**	**278**	**226**	**0.12**	**0.040**	**0.22**	**8.18**	**8.13**	**0.053**

Table D.4: Background data for Option 3 with median and 95th Percentile values.

Year	Month	Dissolved Arsenic (mg/L)	Dissolved Cadmium (mg/L)	Dissolved Chromium (mg/L)	Dissolved Copper (mg/L)	Dissolved Iron (mg/L)	Dissolved Lead (mg/L)	Dissolved Molybdenum (mg/L)	Dissolved Nickel (mg/L)	Dissolved Selenium (mg/L)	Dissolved Silver (mg/L)	Dissolved Zinc (mg/L)
2005	May	0.00029	0.000025	0.00050	0.0042	0.14	0.00025	0.00050	0.0014	0.00050	0.000010	0.0029
	June	0.00046	0.000025	0.00030	0.0023	0.16	0.000064	0.00085	0.0011	0.00050	0.0000050	0.0091
	July	0.00041	0.000025	0.00025	0.0033	0.16	0.000075	0.0011	0.0011	0.00050	0.0000050	0.0030
	August	0.00043	0.000010	0.00050	0.0026	0.27	0.00025	0.00083	0.0014	0.00050	0.000010	0.0029
	September	0.00039	0.0000085	0.00050	0.0027	0.31	0.00025	0.00059	0.0013	0.00050	0.000010	0.0025
	October	0.00044	0.000025	0.00025	0.0029	0.25	0.000025	0.00069	0.0012	0.00050	0.0000050	0.0013
2006	June	0.00051	0.000021	0.00050	0.0017	0.15	0.000094	0.00055	0.0019	0.00039	0.0000050	0.0025
	July	0.00048	0.0000050	0.00076	0.0011	0.083	0.000088	0.0011	0.0012	0.00013		0.00063
	August	0.00035	0.0000050	0.00059	0.0013	0.083	0.000050	0.00088	0.00083	0.00010		0.0024
	September	0.00046	0.000039	0.00070	0.0011	0.10	0.000050	0.00075	0.00067	0.00011		0.0017
	October	0.00035	0.0000050	0.00025	0.00075	0.10	0.000050	0.0010	0.00065	0.00020		0.0015
2007	June	0.00040	0.0000050	0.00060	0.0020	0.050	0.000050	0.0010	0.00025	0.00010		0.0040
	July	0.00050	0.0000050	0.0010	0.0020	0.040	0.000050	0.0010	0.00070	0.00010		0.0060
	August											
	September	0.00040	0.0000050	0.0016	0.0010	0.080	0.000050	0.0020	0.00080	0.00010		0.0020
	October											
2008	April	0.00027	0.000040	0.00097	0.0012	0.030	0.000067	0.0017	0.0010	0.00030		0.0017
	June	0.00010	0.000040	0.0013	0.00075	0.060	0.000050	0.0014	0.0010	0.0013	0.0000050	0.0020
	August	0.00040	0.0000050	0.00060	0.0020	0.070	0.000050	0.0010	0.0010	0.00010	0.000060	0.0060
	September	0.00035	0.0000050	0.0015	0.0030	0.15	0.00088	0.00050	0.0015	0.00075	0.0000050	0.0030
	October	0.00050	0.0000050	0.00050	0.0010	0.078	0.000050	0.00108	0.0010	0.00030	0.0000050	0.0010
2009	April	0.00010	0.000080	0.00060	0.0050	0.070	0.00010	0.00013	0.00050	0.00030	0.0000050	0.0090
	June	0.00030	0.000005	0.00020	0.0020	0.050	0.000050	0.00127	0.0010	0.00030	0.0000050	0.0020
	July											
	August	0.00027	0.000030	0.00067	0.0037	0.073	0.00018	0.00083	0.00083	0.00030	0.0000050	0.0040
	October	0.00030	0.000040	0.00050	0.0012	0.046	0.00010	0.0010	0.00050	0.00020	0.000010	0.0025
2010	April	0.00030	0.000010	0.00047	0.0045	0.14	0.00017	0.00060	0.00083	0.00013	0.0000083	0.0033
	May	0.00038	0.000023	0.00050	0.0021	0.19	0.00010	0.0010	0.0011	0.00022	0.000010	0.0034
	June	0.00028	0.000050		0.00095	0.035	0.00010	0.0016	0.00091	0.00040		0.0050
	July	0.00020	0.000050		0.0020	0.10	0.00010	0.0010	0.0010	0.00040		0.0050
	August	0.00040	0.000050		0.0020	0.11	0.00010	0.0010	0.0010	0.00040		0.0050
	September	0.00050	0.000050		0.0010	0.23	0.00010	0.0010	0.0020	0.00040		0.0050
	October	0.00040	0.0000050	0.00050	0.0012	0.13	0.00010	0.0010	0.0010	0.00010	0.000010	0.0025
2011	April	0.00030	0.000030	0.00050	0.0077	0.15	0.00010	0.00050	0.0010	0.00005	0.000010	0.0050
	May	0.00020	0.000040	0.00050	0.0070	0.17	0.00010	0.00050	0.0010	0.00005	0.000010	0.010
	June	0.00030	0.000020	0.00050	0.0046	0.16	0.00010	0.00075	0.0013	0.000075	0.000010	0.005
	July	0.00070	0.000010	0.00050	0.0031	0.44	0.00010	0.00050	0.0020	0.00010	0.000010	0.0025
	August	0.00073	0.000011	0.00050	0.0028	0.55	0.00010	0.00050	0.0023	0.00010	0.000010	0.0025
	September	0.00070	0.000010	0.00050	0.0016	0.62	0.00010	0.00050	0.0020	0.00010	0.000010	0.0025
	October	0.00060	0.000030	0.00050	0.0012	0.35	0.00010	0.0010	0.0010	0.00010	0.000010	0.0025
	November	0.00040	0.0000050	0.00050	0.0011	0.17	0.00010	0.0013	0.0012	0.00016	0.000010	0.0025
2012	January	0.00041	0.000035	0.00050	0.0011	0.094	0.00010	0.0015	0.00050	0.00043	0.000010	0.0025
	February	0.00033	0.000064	0.00050	0.0154	0.049	0.00024	0.0014	0.00050	0.00034	0.000010	0.0083
	March	0.00039	0.0000050	0.00050	0.0012	0.012	0.00010	0.0011	0.00050	0.00031	0.000010	0.0025
	April	0.00048	0.000023	0.00050	0.0052	0.48	0.00010	0.00050	0.0012	0.000050	0.000010	0.0025
	May	0.00089	0.000021	0.00050	0.0027	1.0	0.00015	0.00050	0.0018	0.000067	0.000010	0.0025
	June	0.00070	0.0000050	0.00050	0.0016	0.56	0.00010	0.0013	0.002	0.00015	0.000010	0.0025
	July	0.00096	0.0000062	0.00050	0.0014	0.73	0.00029	0.0011	0.0015	0.00012	0.000010	0.0025
	August	0.00128	0.0000050	0.00050	0.0014	0.91	0.00010	0.0012	0.0020	0.00011	0.000010	0.0025
	September	0.00061	0.0000050	0.00050	0.0019	0.42	0.00010	0.00050	0.0025	0.000095	0.000010	0.0025
	October	0.00060	0.0000050	0.00050	0.00094	0.60	0.00010	0.0011	0.0015	0.00014	0.000010	0.0025
	November	0.00046	0.000018	0.00050	0.0019	0.10	0.00016	0.00090	0.0013	0.00014	0.000010	0.0025
	December	0.00051	0.0000050	0.00050	0.0023	0.030	0.00010	0.0012	0.0011	0.00045	0.000010	0.0025
2013	April	0.00025	0.0000050	0.00050	0.0011	0.014	0.00010	0.0013	0.00050	0.00043	0.000010	0.0025
	May	0.00033	0.000011	0.00050	0.0036	0.24	0.00010	0.00050	0.0012	0.00005	0.000010	0.0025
	June	0.0011	0.0000085	0.00050	0.0017	0.59	0.00010	0.0013	0.0019	0.00011	0.000010	0.0045
	July	0.0011	0.0000050	0.00050	0.0012	0.66	0.00010	0.0012	0.0018	0.000093	0.000010	0.0025
	August	0.0011	0.0000050	0.00050	0.0012	0.71	0.00010	0.0012	0.0018	0.000093	0.000010	0.0025
	September	0.0012	0.0000056	0.00065	0.0012	0.96	0.00010	0.00083	0.0020	0.000080	0.000010	0.0025
	October	0.00078	0.0000050	0.00050	0.0010	0.65	0.00010	0.00076	0.0017	0.000080	0.000010	0.0025
	November	0.0022	0.000011	0.00077	0.00086	4.8	0.00010	0.0011	0.0025	0.00018	0.000010	0.0025
	December	0.00057	0.0000080	0.00050	0.0012	0.062	0.00010	0.00080	0.00080	0.00015	0.000010	0.0025
Median:		**0.00041**	**0.000010**	**0.00050**	**0.0017**	**0.15**	**0.00010**	**0.0010**	**0.0011**	**0.00015**	**0.000010**	**0.0025**
95th Percentile:		**0.0011**	**0.000050**	**0.0011**	**0.0054**	**0.91**	**0.00025**	**0.0015**	**0.0020**	**0.00050**	**0.000010**	**0.0084**

Table D.5: Annual Medians for Option 3.

Year	Month	DOC (mg/L)	Dissolved Hardness (mg/L)	Total Hardness (mg/L)	TSS (mg/L)	TDS (mg/L)	Ammonia (mg/L)	Nitrite (mg/L)	Nitrate (mg/L)	Lab pH (pH Units)	Field pH (pH Units)	Dissolved Aluminum (mg/L)
2005	May			74	19	118	0.017	0.0011	0.034	7.85		0.043
	June				5.4	160	0.015	0.00067	0.052	7.83		0.026
	July			138	11	173	0.010	0.00050	0.042	7.73		0.020
	August			110	4.9	157	0.029	0.00066	0.010	8.10		0.022
	September			111	58	151	0.012	0.00057	0.025	7.83		0.024
	October			111	5.5	151	0.015	0.00050	0.060	7.64		0.033
	Annual Median			**111**	**8.4**	**154**	**0.015**	**0.00061**	**0.038**	**7.83**		**0.025**
2006	June		110		206	112	0.026	0.0010	0.083	7.71		0.075
	July		120		3.8	184	0.014		0.19	7.86		0.0056
	August		124		2.0	175	0.017		0.075	7.92		0.011
	September		120		4.0	182	0.010		0.053	7.78		0.0079
	October		129		3.0	164	0.052		0.11	7.67		0.0048
	Annual Median		**120**		**3.8**	**175**	**0.017**	**0.0010**	**0.083**	**7.78**		**0.0079**
2007	June	8.2	110	97	28	147	0.025	0.027	0.14	7.84		0.014
	July	14	128	125	5.7	179	0.025	0.025	0.30	7.78		0.011
	August			124	4.7	192	0.025	0.025	0.20	8.04		
	September	11	110	93	14	177	0.025	0.010	0.030	7.98		0.0070
	October			129	6.5	174	0.025	0.14	0.010	7.73		
	Annual Median	**11**	**110**	**124**	**6.5**	**177**	**0.025**	**0.025**	**0.14**	**7.84**		**0.011**
2008	April		187		1.0	268	0.025	0.055	0.010	8.40		0.010
	June		92	92	15	133	0.025	0.058	0.049	7.90		0.013
	August	9.0	123	124	105	167	0.025	0.005	0.143	7.92		0.014
	September	18			16	160	0.025	0.005	0.018	7.85		0.041
	October	11	123		5.7	183	0.025			7.84		0.0090
	Annual Median	**11**	**123**	**108**	**15**	**167**	**0.025**	**0.030**	**0.033**	**7.90**		**0.013**
2009	April	18	20		1.0	52				6.84	7.60	0.053
	June	9.2	93		32	156				7.90	7.74	0.013
	July		137		4.0	184				8.07		
	August	12	96		9.8	149			0.18	7.79	7.72	0.025
	October	8.3	125	141	2.8	130	0.010	0.0025	0.14	8.02	8.24	0.007
	Annual Median	**10**	**96**	**141**	**4.0**	**149**	**0.010**	**0.0025**	**0.16**	**7.90**	**7.73**	**0.019**
2010	April	21	36	50	2.5	89	0.016	0.0038	0.015	7.64	7.56	0.053
	May	11	83	88	67	119	0.034	0.0065	0.070	8.02	7.64	0.029
	June	7.6	113	121	63	143	0.059	0.0073	0.14	8.17	7.49	0.012
	July	16	93	106	53	150	0.027	0.0025	0.070	8.06	8.02	0.019
	August	13	118	116		170					7.91	0.011
	September	17	107	104	11	210	0.039	0.0025	0.030	7.98	8.02	0.016
	October	13	111	126	3.0	150	0.069	0.0025	0.070	8.10	7.85	0.011
	Annual Median	**13**	**107**	**106**	**32**	**150**	**0.037**	**0.0031**	**0.070**	**8.04**	**7.85**	**0.016**
2011	April	23	43	43	0.50	72	0.010	0.0025	0.010	7.41		0.057
	May	22	32	33	7.0	40		0.0025	0.010	7.11		0.076
	June	17	66	69	29	91	0.012	0.0025	0.040	7.65		0.042
	July	21	77	88	150	120	0.016	0.0070	0.050	7.76		0.034
	August	21	82	91	149	130	0.012	0.0036	0.053	7.86		0.039
	September	13	118	119	1.0	170	0.025	0.0025	0.050	8.03	7.92	0.024
	October	12	110	113	2.0	150	0.019	0.0025	0.11	8.07	7.65	0.013
	November	7.7	148	144	27	156	0.021	0.0025	0.14	7.96		0.0076
	Annual Median	**19**	**79**	**89**	**17**	**125**	**0.016**	**0.0025**	**0.050**	**7.81**	**7.79**	**0.036**
2012	January	5.3	169	158	2.0	226	0.0050	0.0025	0.32	8.13		0.0053
	February	6.1	165	160	2.0	216	0.0060	0.0025	0.29	7.99	7.16	0.021
	March	5.2	161	159	3.2	176	0.0067	0.0025	0.20	8.17	6.72	0.0031
	April	14	39	37	11.2	100	0.0025	0.010	0.010	7.63	7.43	0.050
	May	14	52	126	1158	100	0.093	0.0055	0.029	7.07	7.50	0.053
	June	16	92	105	165	128	0.18	0.015	0.083	8.12	7.59	0.027
	July	13	121	162	630	158	0.134	0.010	0.11	8.15	7.86	0.031
	August	11	128	175	700	178	0.045	0.0097	0.10	8.16	8.08	0.030
	September	15	102	107	33	136	0.0099	0.0025	0.056	8.09	7.97	0.020
	October	11	129	130	27	158	0.022	0.0025	0.084	8.05	8.10	0.011
	November	8.5	156	154	2.2	188	0.026	0.0025	0.11	7.93	6.69	0.0079
	December	10	223	206	0.50	262	0.025	0.0025	0.063	8.13	7.97	0.0061
	Annual Median	**11**	**128**	**156**	**19**	**167**	**0.024**	**0.0025**	**0.092**	**8.10**	**7.59**	**0.021**
2013	April	4.6	192	198	5.8	246	0.0082	0.0025	0.13	8.27	8.18	0.0031
	May	19	32	35	34	77	0.098	0.0025	0.010	7.53	7.90	0.051
	June	13	121	151	278	168	0.16	0.011	0.17	8.09	8.05	0.034
	July	9.1	118	147	179	163	0.11	0.0092	0.091	8.08	7.99	0.019
	August	13	131	154	218	179	0.080	0.0052	0.097	8.12	7.92	0.021
	September	12	129	149	53	196	0.048	0.0037	0.10	7.98	7.80	0.022
	October	12	130	133	40	193	0.069	0.0025	0.11	8.18	7.55	0.014
	November	16	177	184	19	185	0.11	0.010	0.12	8.08	7.61	0.015
	December	8.0	164	186	2.1	207	0.016	0.0025	0.11	8.18	6.90	0.0038
	Annual Median	**12**	**130**	**151**	**40**	**185**	**0.080**	**0.0037**	**0.11**	**8.09**	**7.90**	**0.019**

Table D.5: Annual Medians for Option 3.

Year	Month	Dissolved Arsenic (mg/L)	Dissolved Cadmium (mg/L)	Dissolved Chromium (mg/L)	Dissolved Copper (mg/L)	Dissolved Iron (mg/L)	Dissolved Lead (mg/L)	Dissolved Molybdenum (mg/L)	Dissolved Nickel (mg/L)	Dissolved Selenium (mg/L)	Dissolved Silver (mg/L)	Dissolved Zinc (mg/L)
2005	May	0.00029	0.000025	0.00050	0.0042	0.14	0.00025	0.00050	0.0014	0.00050	0.000010	0.0029
	June	0.00046	0.000025	0.00030	0.0023	0.16	0.000064	0.00085	0.0011	0.00050	0.0000050	0.0091
	July	0.00041	0.000025	0.00025	0.0033	0.16	0.000075	0.0011	0.0011	0.00050	0.0000050	0.0030
	August	0.00043	0.000010	0.00050	0.0026	0.27	0.00025	0.00083	0.0014	0.00050	0.000010	0.0029
	September	0.00039	0.0000085	0.00050	0.0027	0.31	0.00025	0.00059	0.0013	0.00050	0.000010	0.0025
	October	0.00044	0.000025	0.00025	0.0029	0.25	0.000025	0.00069	0.0012	0.00050	0.0000050	0.0013
	Annual Median	**0.00042**	**0.000025**	**0.00040**	**0.0028**	**0.21**	**0.00016**	**0.00076**	**0.0013**	**0.00050**	**0.0000075**	**0.0029**
2006	June	0.00051	0.000021	0.00050	0.0017	0.15	0.000094	0.00055	0.0019	0.00039	0.0000050	0.0025
	July	0.00048	0.0000050	0.00076	0.0011	0.083	0.000088	0.0011	0.0012	0.00013		0.00063
	August	0.00035	0.0000050	0.00059	0.0013	0.083	0.000050	0.00088	0.00083	0.00010		0.0024
	September	0.00046	0.000039	0.00070	0.0011	0.10	0.000050	0.00075	0.00067	0.00011		0.0017
	October	0.00035	0.0000050	0.00025	0.00075	0.10	0.000050	0.0010	0.00065	0.00020		0.0015
	Annual Median	**0.00046**	**0.0000050**	**0.00059**	**0.0011**	**0.10**	**0.000050**	**0.00088**	**0.00083**	**0.00013**	**0.0000050**	**0.0017**
2007	June	0.00040	0.0000050	0.00060	0.0020	0.050	0.000050	0.0010	0.00025	0.00010		0.0040
	July	0.00050	0.0000050	0.0010	0.0020	0.040	0.000050	0.0010	0.00070	0.00010		0.0060
	August											
	September	0.00040	0.0000050	0.0016	0.0010	0.080	0.000050	0.0020	0.00080	0.00010		0.0020
	October											
	Annual Median	**0.00040**	**0.0000050**	**0.0010**	**0.0020**	**0.050**	**0.000050**	**0.0010**	**0.00070**	**0.00010**		**0.0040**
2008	April	0.00027	0.000040	0.00097	0.0012	0.030	0.000067	0.0017	0.0010	0.00030		0.0017
	June	0.00010	0.000040	0.0013	0.00075	0.060	0.000050	0.0014	0.0010	0.0013	0.0000050	0.0020
	August	0.00040	0.0000050	0.00060	0.0020	0.070	0.000050	0.0010	0.0010	0.00010	0.000060	0.0060
	September	0.00035	0.0000050	0.0015	0.0030	0.15	0.00088	0.00050	0.0015	0.00075	0.0000050	0.0030
	October	0.00050	0.0000050	0.00050	0.0010	0.078	0.000050	0.00108	0.0010	0.00030	0.0000050	0.0010
	Annual Median	**0.00035**	**0.0000050**	**0.0010**	**0.0012**	**0.070**	**0.000050**	**0.0011**	**0.0010**	**0.00030**	**0.0000050**	**0.0020**
2009	April	0.00010	0.000080	0.00060	0.0050	0.070	0.00010	0.00013	0.00050	0.00030	0.0000050	0.0090
	June	0.00030	0.000005	0.00020	0.0020	0.050	0.000050	0.00127	0.0010	0.00030	0.0000050	0.0020
	July											
	August	0.00027	0.000030	0.00067	0.0037	0.073	0.00018	0.00083	0.00083	0.00030	0.0000050	0.0040
	October	0.00030	0.000040	0.00050	0.0012	0.046	0.00010	0.0010	0.00050	0.00020	0.000010	0.0025
	Annual Median	**0.00028**	**0.000035**	**0.00055**	**0.0028**	**0.060**	**0.00010**	**0.00092**	**0.00067**	**0.00030**	**0.0000050**	**0.0033**
2010	April	0.00030	0.000010	0.00047	0.0045	0.14	0.00017	0.00060	0.00083	0.00013	0.0000083	0.0033
	May	0.00038	0.000023	0.00050	0.0021	0.19	0.00010	0.0010	0.0011	0.00022	0.000010	0.0034
	June	0.00028	0.000050		0.00095	0.035	0.00010	0.0016	0.00091	0.00040		0.0050
	July	0.00020	0.000050		0.0020	0.10	0.00010	0.0010	0.0010	0.00040		0.0050
	August	0.00040	0.000050		0.0020	0.11	0.00010	0.0010	0.0010	0.00040		0.0050
	September	0.00050	0.000050		0.0010	0.23	0.00010	0.0010	0.0020	0.00040		0.0050
	October	0.00040	0.0000050	0.00050	0.0012	0.13	0.00010	0.0010	0.0010	0.00010	0.000010	0.0025
	Annual Median	**0.00038**	**0.000050**	**0.00050**	**0.0020**	**0.13**	**0.00010**	**0.0010**	**0.0010**	**0.00040**	**0.000010**	**0.0050**
2011	April	0.00030	0.000030	0.00050	0.0077	0.15	0.00010	0.00050	0.0010	0.00005	0.000010	0.0050
	May	0.00020	0.000040	0.00050	0.0070	0.17	0.00010	0.00050	0.0010	0.00005	0.000010	0.010
	June	0.00030	0.000020	0.00050	0.0046	0.16	0.00010	0.00075	0.0013	0.000075	0.000010	0.005
	July	0.00070	0.000010	0.00050	0.0031	0.44	0.00010	0.00050	0.0020	0.00010	0.000010	0.0025
	August	0.00073	0.000011	0.00050	0.0028	0.55	0.00010	0.00050	0.0023	0.00010	0.000010	0.0025
	September	0.00070	0.000010	0.00050	0.0016	0.62	0.00010	0.00050	0.0020	0.00010	0.000010	0.0025
	October	0.00060	0.000030	0.00050	0.0012	0.35	0.00010	0.0010	0.0010	0.00010	0.000010	0.0025
	November	0.00040	0.0000050	0.00050	0.0011	0.17	0.00010	0.0013	0.0012	0.00016	0.000010	0.0025
	Annual Median	**0.00050**	**0.000016**	**0.00050**	**0.0029**	**0.26**	**0.00010**	**0.00050**	**0.0012**	**0.00010**	**0.000010**	**0.000010**
2012	January	0.00041	0.000035	0.00050	0.0011	0.094	0.00010	0.0015	0.00050	0.00043	0.000010	0.0025
	February	0.00033	0.000064	0.00050	0.0154	0.049	0.00024	0.0014	0.00050	0.00034	0.000010	0.0083
	March	0.00039	0.0000050	0.00050	0.0012	0.012	0.00010	0.0011	0.00050	0.00031	0.000010	0.0025
	April	0.00048	0.000023	0.00050	0.0052	0.48	0.00010	0.00050	0.0012	0.000050	0.000010	0.0025
	May	0.00089	0.000021	0.00050	0.0027	1.0	0.00015	0.00050	0.0018	0.000067	0.000010	0.0025
	June	0.00070	0.0000050	0.00050	0.0016	0.56	0.00010	0.0013	0.002	0.00015	0.000010	0.0025
	July	0.00096	0.0000062	0.00050	0.0014	0.73	0.00029	0.0011	0.0015	0.00012	0.000010	0.0025
	August	0.00128	0.0000050	0.00050	0.0014	0.91	0.00010	0.0012	0.0020	0.00011	0.000010	0.0025
	September	0.00061	0.0000050	0.00050	0.0019	0.42	0.00010	0.00050	0.0025	0.000095	0.000010	0.0025
	October	0.00060	0.0000050	0.00050	0.00094	0.60	0.00010	0.0011	0.0015	0.00014	0.000010	0.0025
	November	0.00046	0.000018	0.00050	0.0019	0.10	0.00016	0.00090	0.0013	0.00014	0.000010	0.0025
	December	0.00051	0.0000050	0.00050	0.0023	0.030	0.00010	0.0012	0.0011	0.00045	0.000010	0.0025
	Annual Median	**0.00056**	**0.0000056**	**0.00050**	**0.0018**	**0.45**	**0.00010**	**0.0011**	**0.0014**	**0.00014**	**0.000010**	**0.0025**
2013	April	0.00025	0.0000050	0.00050	0.0011	0.014	0.00010	0.0013	0.00050	0.00043	0.000010	0.0025
	May	0.00033	0.000011	0.00050	0.0036	0.24	0.00010	0.00050	0.0012	0.00005	0.000010	0.0025
	June	0.0011	0.0000085	0.00050	0.0017	0.59	0.00010	0.0013	0.0019	0.00011	0.000010	0.0045
	July	0.0011	0.0000050	0.00050	0.0012	0.66	0.00010	0.0012	0.0018	0.000093	0.000010	0.0025
	August	0.0011	0.0000050	0.00050	0.0012	0.71	0.00010	0.0012	0.0018	0.000093	0.000010	0.0025
	September	0.0012	0.0000056	0.00065	0.0012	0.96	0.00010	0.00083	0.0020	0.000080	0.000010	0.0025
	October	0.00078	0.0000050	0.00050	0.0010	0.65	0.00010	0.00076	0.0017	0.000080	0.000010	0.0025
	November	0.0022	0.000011	0.00077	0.00086	4.8	0.00010	0.0011	0.0025	0.00018	0.000010	0.0025
	December	0.00057	0.0000080	0.00050	0.0012	0.062	0.00010	0.00080	0.00080	0.00015	0.000010	0.0025
	Annual Median	**0.0011**	**0.0000056**	**0.00050**	**0.0012**	**0.65**	**0.00010**	**0.0011**	**0.0018**	**0.000093**	**0.000010**	**0.0025**

Table D.6: Background water quality summary statistics for Option 2 (Grouped by Week) representing lower Minto Creek (Minnow 2016) [1]

Analytes	Background Concentrations	
	Individual Data Point Evaluator (95th percentile)	Central Tendency Evaluator (95th percentile of annual medians)
Ammonia (mg/L)	0.12	0.058
Nitrite (mg/L)	0.038	0.034
Nitrate (mg/L)	0.22	0.14
Dissolved Aluminum (mg/L)	0.056	0.027
Dissolved Arsenic (mg/L)	0.0013	0.00077
Dissolved Cadmium (mg/L)	0.000050	0.000046
Dissolved Chromium (mg/L)	0.00130	0.00108
Dissolved Copper (mg/L)	0.0051	0.0025
Dissolved Iron (mg/L)	0.95	0.55
Dissolved Lead (mg/L)	0.000276	0.000138
Dissolved Molybdenum (mg/L)	0.00172	0.00114
Dissolved Nickel (mg/L)	0.0024	0.0017
Dissolved Selenium (mg/L)	0.00050	0.00046
Dissolved Silver (mg/L)	0.000010	0.000010
Dissolved Zinc (mg/L)	0.0060	0.0046

[1] For contaminants of potential concern (COPCs)

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Executive Summary

Passive water treatment is desired in closure at the Minto Mine, with a focus on the constituents of cadmium (Cd), copper (Cu), and selenium (Se). To address this goal, a constructed wetland treatment system (CWTS) is being designed in a site-specific manner, through a phased approach.

This document outlines foundational scientific information used to select the passive water treatment technology and develop conceptual designs for the Minto Mine. This document then proceeds through to the preliminary CWTS design which is based on findings from phases that included: information gathering, site assessment, off-site pilot-scale testing, and preliminary on-site demonstration-scale confirmation and optimization.

This report specifically considers water treatment by a CWTS within the footprint of the water storage pond at the Minto Mine. The preliminary CWTS design described herein is a surface flow wetland, designed to create and maintain anaerobic conditions in the sediment to sequester cadmium and copper in sulphide mineral form, and selenium in a low bioavailable elemental form. The phased plan for site-specific design and optimization has demonstrated that these elements can be successfully and sustainably treated at the Minto Mine through passive treatment such as a CWTS. A CWTS of this type can be brought online effectively by having more rigorous monitoring and adjustments during an initial commissioning period, followed by progressive decreases in monitoring and maintenance requirements through early- and long-term operation. Conceptual monitoring and maintenance schedules are also described herein.

1. Introduction

The Minto Mine, operated by Capstone Mining Corp., is located 240 km northwest of Whitehorse on the west side of the Yukon River. The Minto property lies within the eastern part of the Dawson Range, with elevations from 700 to 1,000 m; the landscape has rounded mountains intersected by broad valleys and drainages that are part of the Yukon River watershed.

The Minto Mine has been in commercial operation since October 2007 and the deposits being mined are copper sulphide mineralized zones. Surface and groundwater water quality is a key consideration in the evaluation of potential effects of mining and mineral development projects and changes to water quality parameters have the potential to affect aquatic and human use of water resources. A Reclamation and Closure Plan (RCP) is required under both the Water Licence and the Quartz Mining Licence. The RCP is intended to address the long-term physical and chemical stability of the site and closure of the proposed features and disturbances associated with the mine. As a part of the RCP, a Constructed Wetland Treatment System (CWTS) is being designed, evaluated, and optimized for water treatment at closure through a phased program (Minto Phase V/VI Expansion Project, YOR Project Number 2013-0100).

The purpose of this report is to describe the design considerations and conceptual monitoring and maintenance associated with a CWTS in the Water Storage Pond (WSP) area of the Minto Mine (Minto) in closure. The scope of this report resides with maintaining performance and treatment capacity, while engineering aspects to maintain hydrology (which in turn, is necessary for treatment) are addressed in Appendix A (SRK, 2016b).

2. Design

2.1. Treatment Technologies

Passive treatment systems are proposed as key mitigation measures for the achievement of acceptable water quality from the Minto site in the closure condition (Access, June 2013).

Passive treatment as a remediation method for mine-impacted water can be a sustainable method used during post-closure of a mine, as they generally involve significantly less direct capital costs, as well as lower operations and maintenance costs, when compared to traditional active treatment options (Kilbourn Inc., 1999).

A detailed review of the site-wide water balance confirmed that a significant portion of the mining impacted seepage, surface flow and contaminant load could be passively intercepted at a few specific areas which have sufficient space and topography for the incorporation of a wide variety of passive treatment systems (PTS) such as constructed wetland treatment systems (CWTS).

The various potential passive treatment technologies were evaluated and ranked in order to determine which passive treatment technologies held the greatest promise for incorporation in Minto water management and closure planning. A guidance document to help short list and then select an appropriate method for managing mine waste sites was created by The Interstate Technology and Regulatory Council (ITRC), a body of socio- economic and environmental regulators, industry, federal government, and stakeholders that work towards innovative environmental decision making (ITRC, 2010).

ITRC's "Mining Waste Treatment Technology Selection" guidance document (and associated literature reviews) is an interactive and iterative web-based decision tree which provides a systematic framework that can be used to identify a short list of appropriate technologies (ITRC, 2010). The ITRC decision tree framework was used for identifying the range of potential passive treatment technologies which may be applicable at the Minto Mine because it provides a straightforward and transparent method for selecting treatment technologies. It also provides legitimacy in the selection process by using an internationally recognized method that was created specifically for mining waste sites.

Following the identification of potential passive treatment technologies, a screening level evaluation and ranking of the potential passive treatment technologies was conducted. Some of the key considerations evaluating the various treatment technologies included:

- Work acceptably in cold weather;
- Sustainable in the long-term;
- Minimal/low intensity active maintenance scenario;
- Suitable for addressing Minto's specific constituents of potential concern (i.e., Cu, Cd, Se);
- Robust technology with adequate precedent of successful application;

- Amenable to collaboration with SFN and consistent with their aspirations for long-term employment and future land use;
- Cost effective; and
- Amenable to incorporation in an Adaptive Management Plan and modifying/expanding if initial design underperforms.

Based on key site-specific considerations, CWTS were deemed promising and were selected for further investigation at Minto. This report specifically considers water treatment by a CWTS within the footprint of the water storage pond.

2.2. Design Basis

PTS such as CWTS rely on transfers (e.g., sorption, filtration) and transformations (e.g., geochemical and biogeochemical reactions) to remove constituents from water (Haakensen et al., 2015; Rodgers and Castle, 2008). These processes require no or infrequent (periodic) electricity, amendment addition, substrate replacement, or flow management. Moreover, these processes can be targeted through specific design aspects, therefore identifying the desired processes and mechanisms to promote their activity and enhance their robustness promotion forms the foundation of the design basis of a CWTS.

Once CWTS have undergone commissioning, they can be relatively self-sustaining, operationally passive (or remotely operated), with minimal periodic maintenance or management required (e.g., addition of sources of carbon), and therefore can provide effective low-cost water treatment in remote environments. This, coupled with the ability to work in the absence of power or the need for active management, suggests that PTS are suitable as a component of a long-term solution to post-closure management of mine-influenced water quality.

The following sections discuss treatment mechanisms for constituents of concern at the Minto Mine.

2.2.1. Transfers

Some of the simplest treatment mechanisms within a CWTS are transfers. These include sorption and filtration, which are not greatly affected by the temperature of the water. While facilitating treatment in colder temperatures, the transfer sites need to be renewed as they are only present in a finite abundance. Transfer sites are renewed by the vegetation, which aids filtration (by submerged vegetation and aquatic roots), and provides sorption sites (on live and decomposing vegetation). Transfer sites are particularly abundant in aquatic mosses. Aquatic plants were selected for the Minto CWTS based on being obligate wetland plants, with a high generation of organic matter, and ideally being peat forming. Such plants were identified at the Minto site during the 2013 CWTS site assessment, with *Carex aquatilis* (known as Aquatic Sedge or Water Sedge) and aquatic bryophytes (aquatic mosses, capable of growing fully submerged in water) being selected as the most promising candidates, (Contango, March 2014) and confirmed through off-site pilot-scale testing (Contango, November 2014).

However, once the constituents have been transferred from one location (water) to another (soils, vegetation) within the wetland, they still need to be transformed to a more stable and less bioavailable form, otherwise known as a coupled biogeochemical reaction. As the CWTS design aspects associated with transfers are fairly generic, the remainder of the design basis was formed around the biogeochemical transformations needed for long-term treatment and sequestration of constituents.

2.2.2. Transformations

Transformations in a CWTS are largely governed by biogeochemical reactions are catalyzed by microbes. Specific microbes (and therefore biogeochemical pathways) can be encouraged through the environmental conditions created by the design of a CWTS. For example, metals can be removed from water by forming insoluble compounds with sulphides. These sulphides are generated by bacteria (sulphide-producing bacteria) that use organic carbon sources (e.g., wood chips, plant matter, methanol) to reduce sulphate in the water under specific oxidation-reduction conditions (Figure 1 and Table 1). This mechanism can be used to treat multiple metals, however, some metals and metalloids will be more readily removed than others as a sulphide mineral, while others can be directly treated through oxidation or reductive processes (Table 1). Therefore, the constituents targeted for treatment at the Minto mine (cadmium, copper, and selenium) are individually addressed in the sections below.

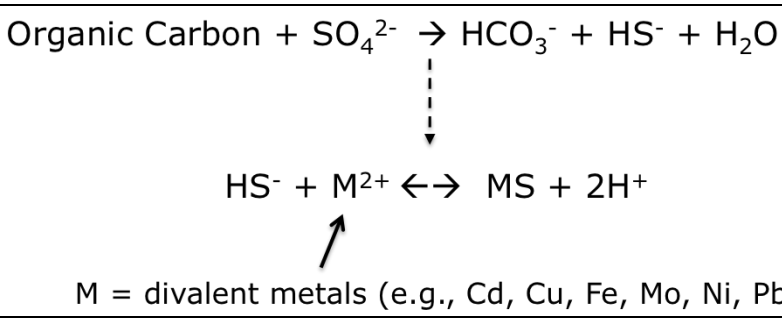


Figure 1 – Example equation of sulphate reduction and metals treatment.
The equation is not balanced and varies based on carbon source and pH. The equation shows the creation of alkalinity (HCO_3^-) and sulphides (HS^-), the latter which can then form insoluble metal sulphide complexes.

Table 1 – ORP values for various biochemical reactions.

Biochemical Reaction	ORP (mV)
Nitrification	+100 to +350
cBOD degradation with free molecular oxygen	+50 to +250
Biological phosphorus removal	+25 to +250
Denitrification	+50 to -50
Selenium reduction	-50 to -200
Sulphide formation	-50 to -250
Biological phosphorus release	-100 to -250
Acid formation (fermentation)	-100 to -225
Methane production	-175 to -400
Adapted from Gerardi, 2010.	

2.2.3. Copper treatment mechanisms

Copper treatment can be achieved in several ways, through oxidizing or reducing geochemical and biogeochemical reactions. When treated from water under appropriately designed reducing conditions, copper will form insoluble copper sulphides (Figure 1, Figure 2 and Figure 3). In contrast, in oxidizing conditions, copper can be removed from water as insoluble hydroxides, carbonates, or iron and manganese oxides (Figure 2).

Several copper treatment mechanisms can occur abiotically (i.e., geochemical reaction) given the right conditions and constituents in water (e.g., oxidizing and co-precipitation with iron). In contrast, formation of copper sulphides is generally biologically driven. It requires the production of sulphides by microbes under anaerobic (i.e., depleted dissolved oxygen) and reducing conditions (-50 to -250 mV; Table 1), with a carbon source to sustain the reaction. Although literature provides a range of -50 to -250mV for sulphate reduction, pilot-scale testing for the Minto CWTS, indicated the site-specific targeted soil redox is between -100 and -250 mV.

Within a CWTS, copper can be removed by both oxidizing and reducing conditions, oxidizing associated with actively growing aquatic mosses, and reduction associated with sediments and decaying mosses. These processes are often coupled, such that the long-term sequestration of the copper in the CWTS is as a copper sulphide. Some forms of mineralized copper are more stable and resistant to resolubilization, and the stability of these minerals is reflected by a K_{sp} value (Table 2 and Table 3). Generally, the minerals that have been catalyzed by microbiological activity cannot spontaneously resolubilize under conditions periodically encountered in CWTS. Additionally, the CWTS forms other minerals, such as amorphous FeS which serves as a buffer to re-oxidation of minerals such as copper sulphides (Table 2). For these reasons, reductive mineralization of copper sulphides was chosen as the preferred treatment pathway for copper in this CWTS.

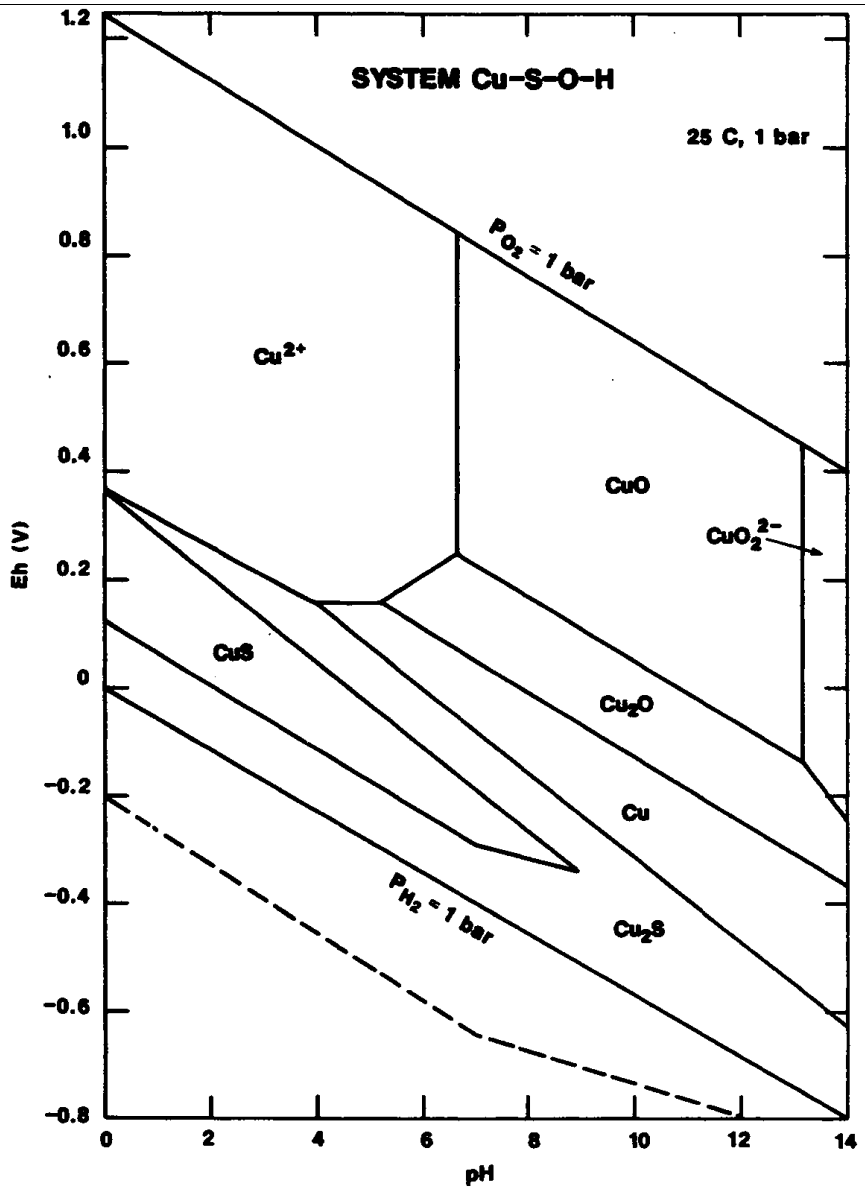


Figure 2 – Eh-pH diagram for part of the system Cu-S-O-H.
See Table 3 for thermodynamic data (Brookins 1988).

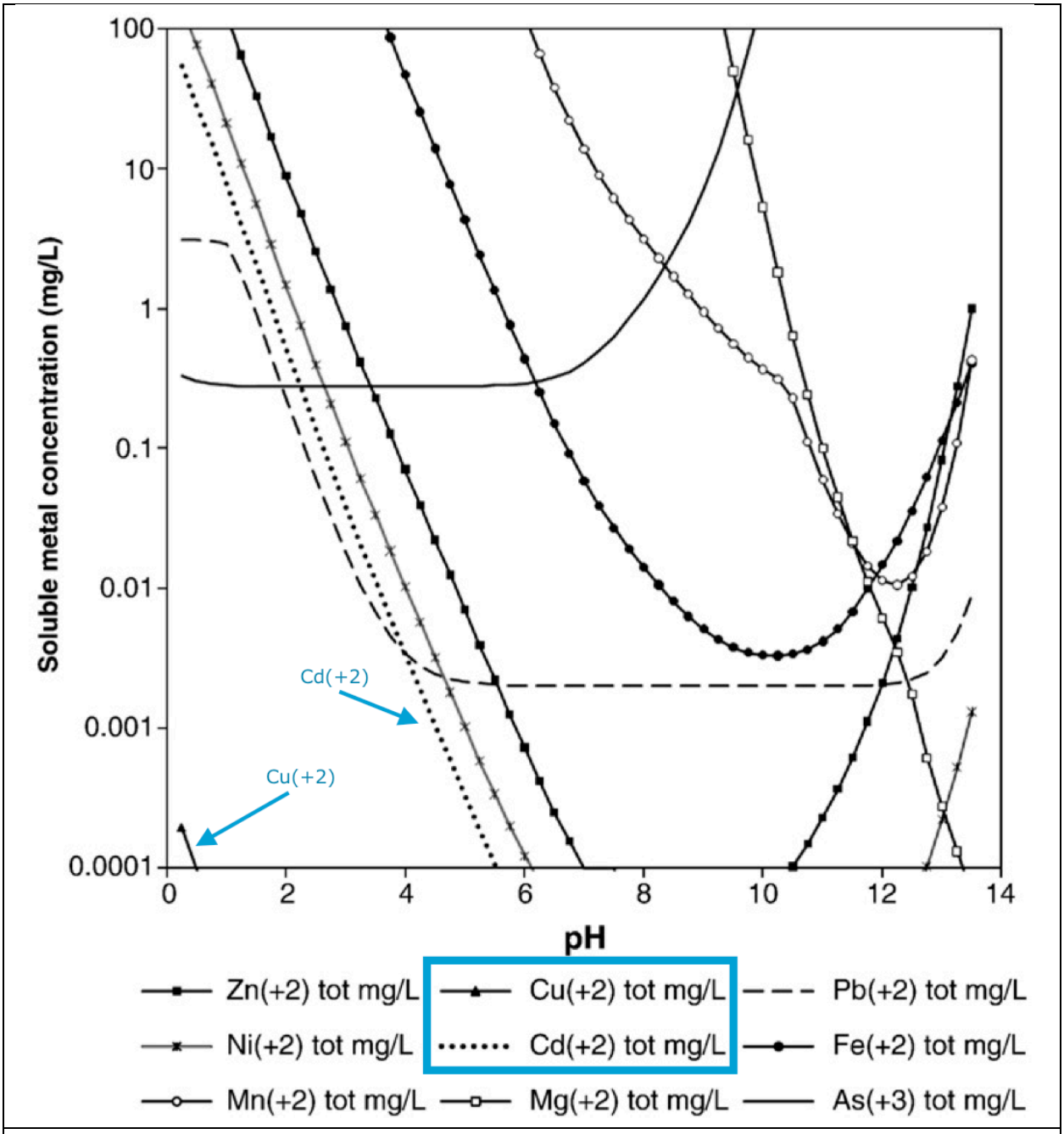


Figure 3 – pH dependence of metal sulphide solubility.
Copper- and cadmium-sulphide solubility indicated with blue arrows and box. Adapted from Lewis (2010).

Table 2 – Solubility of sulphide minerals.

Metal Sulphide	Log K_{sp} [1]
CdS	-25.8
CuS	-36.1
Cu_2S	-47.7
FeS	-18.1

[1] Solubility product (K_{sp}). Table adapted from Jackson (1986).

Table 3 – Thermodynamic data for copper.

Species State	ΔG^0_f(kcal/gfw)
$Cu_{(c)}$	0.00
$Cu^+_{(aq)}$	+11.94
$Cu^{2+}_{(aq)}$	+15.65
$Cu_2S_{(c)}$	-20.60
$CuS_{(c)}$	-12.81
$Cu_2O_{(c)}$	-34.98
$CuO_{(c)}$	-31.00
$CuO_{2\,(aq)}^{2-}$	-43.88
$Cu_2(CO_3)(OH)_{2\ (c)}$	-213.58
$Cu_3(CO_3)_2(OH)_{2\ (c)}$	-314.29

Adapted from Brookins, 1988; aq – aqueous (dissolved/soluble), c – crystalline (solid).

2.2.4. Cadmium treatment mechanisms

Passive treatment mechanisms for cadmium are the same as copper, with precipitation as sulphides, hydroxides, carbonates, and iron and manganese oxides as means for removal depending on water chemistry and conditions (Figure 1, Figure 3, Figure 4, and Table 4). Both sulphides and iron and manganese oxides will achieve the ultralow concentrations desired for the Minto CWTS. Based on site-specific considerations and water chemistry at the Minto Mine, cadmium treatment is targeted to be removed in the most stable and low-bioavailable form, and therefore will be removed by the same mechanism as copper, by precipitation as a metal sulphide (cadmium sulphide, CdS; Table 2).

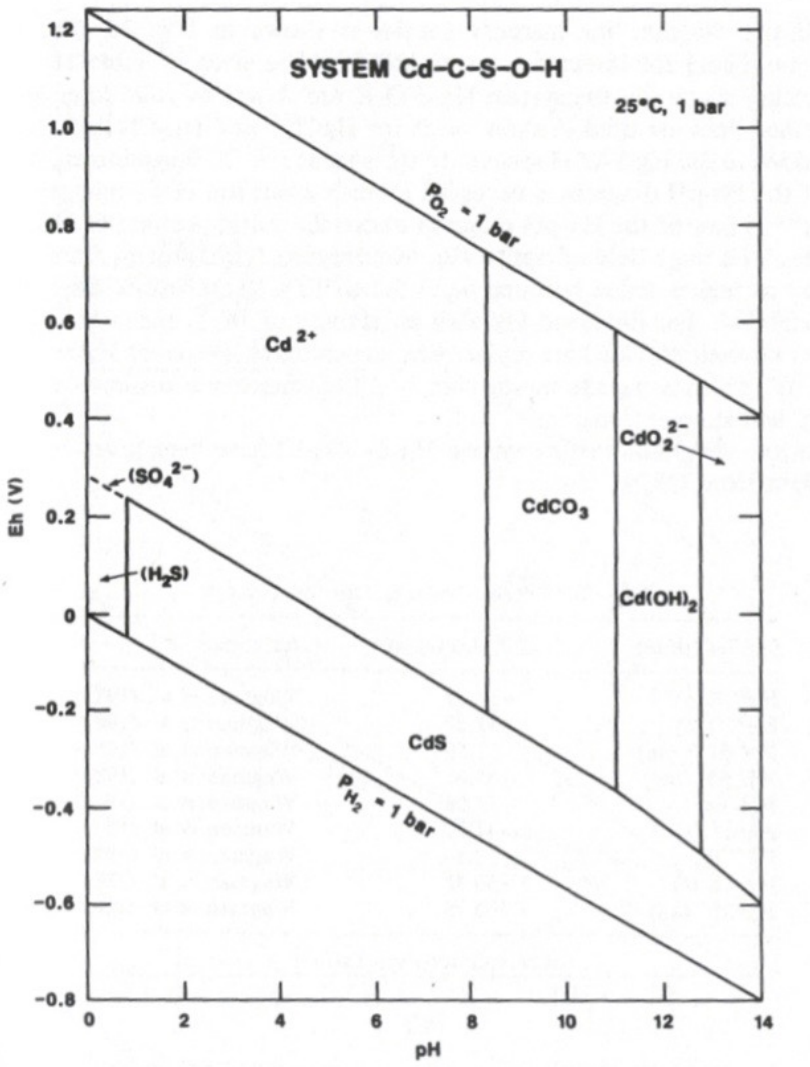


Figure 4 – Eh-pH diagram for cadmium.

Table 4 – Thermodynamic data for cadmium.

Species State	ΔG^0_f(kcal/gfw)
$Cd^{2+}{}_{(aq)}$	-18.55
$Cd(OH)_{2\ (c)}$	-113.19
$CdS_{\ (c)}$	-37.40
$CdO_2^{2-}{}_{(aq)}$	-67.97
$CdCO_{3\ \ (c)}$	-160.00
$CdSO_{4\ \ (c)}$	-196.33
$CdO_{\ (c)}$	-54.59

Adapted from Brookins (1988); aq – aqueous (dissolved/soluble), c – crystalline (solid).

2.2.5. Selenium treatment mechanisms

The mechanism for selenium treatment is more direct, through microbially catalyzed dissimilatory selenium reduction. This is performed directly by bacteria acting on the selenium. The most common forms of aqueous Se are selenite and selenate, with valence states of Se(IV) and Se(VI), respectively. When these are reduced in a wetland to an insoluble state, Se precipitates as elemental Se(0) (Figure 5), which is the lowest bioavailability (Chapman et al., 2010).

While selenium can be directly treated through microbial reduction in a CWTS, there are constituents that can interfere with this process. Most notably, nitrate has a high affinity for available electrons and is preferentially reduced before other compounds including selenium and sulphate. Nitrate especially interferes with selenium reduction, as it may be performed by the same metabolic pathways in the same bacteria (Oremland et al., 1989 and Oremland et al., 1990). While it is known that over time through closure nitrate concentrations will decrease as the source (blasting residue) has ceased, during early closure nitrate may affect the CWTS treatment capacity for selenium. As such, the pilot-scale CWTSs were subjected to a 6-week trial of synthetic water that mimicked an early closure scenario, with nitrate concentrations based on predicted maximum worst-case ammonia and nitrate concentrations. This was followed by a second phase of testing with depleted nitrate concentrations to simulate the chemistry expected in long-term closure (Contango, November 2014).

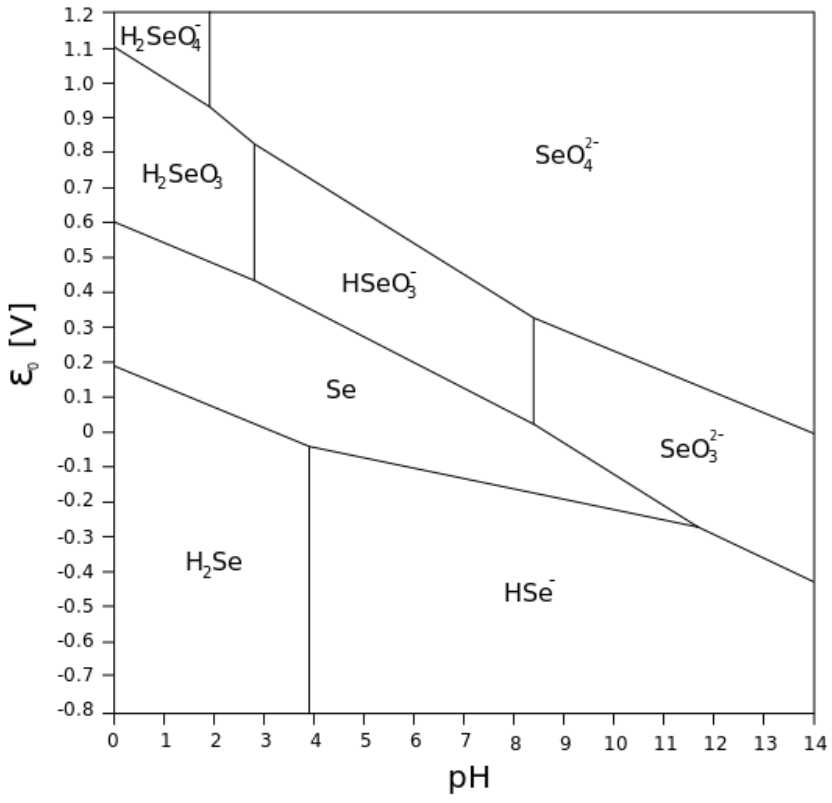


Figure 5 – Eh-pH diagram for selenium.

2.3. Design selection

There are numerous layouts and configurations that can be implemented for treatment wetlands, with varied hydrology, performance parameters, and operation and maintenance requirements. The most passive design was chosen as the goal of final mine site closure is to ensure long-term physical and geochemical stability and to minimize reliance on long-term active treatment. To meet the requirements for water treatment with minimal intervention, the selected passive treatment wetland design is one where there is no operational management necessary, and only minimal periodic maintenance is required, which could be performed by manpower (i.e., without machinery). Based on these guiding objectives, the selected configuration at the Minto Mine is a horizontal surface flow treatment wetland (Figure 6).

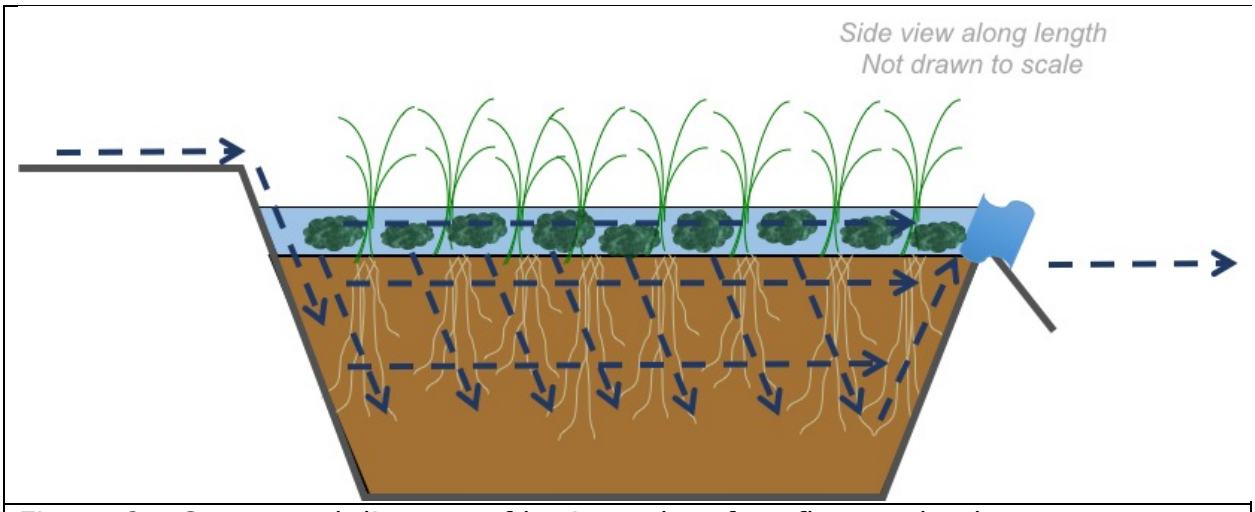


Figure 6 – Conceptual diagram of horizontal surface flow wetland.
Dotted arrows show flow path through wetland and into substrate and root zone. Vegetation in this example includes both emergent macrophytes (e.g., *Carex aquatilis* known as aquatic sedge), and bryophytes (aquatic mosses).

2.3.1. Plants

For the CWTS, *Carex aquatilis* (aquatic sedge) was chosen, as it is readily available at the Minto site and was one of the first plants to colonize cleared areas, indicating that it may be a good candidate for quickly establishing the CWTS (Contango, March 2014). Moss was also used in pilot-scale CWTS designs, as locations at Minto with shallower water had *Carex* and Moss coexisting. The demonstration-scale CWTS was planted with *C. aquatilis* and mosses, as was determined by pilot-scale testing as being the best design for this site and application (Contango, November 2014).

2.3.2. Substrate

The recommended soil for the CWTS is sand, with approximately 2-7% by volume as organic material (e.g., woodchips, straw). In the pilot-scale systems (Contango, November 2014), this resulted in a total organic carbon (TOC) content of 0.2-0.6% (the sand itself was at 0.1% TOC prior to adding amendment), and it was found that a higher concentration would likely decrease the duration of the commissioning

period. In this composition, the sand allows for circulation of water through the root zone of the plants, while the organic matter initiates reductive microbial processes that in future years will be sustained by decaying plant matter. Additionally, sandy soil will support weight while wet for planting the CWTS.

2.3.3. Water Depth

The recommended water depth for the CWTS is based on plant selection. Once plants are well established and showing signs of colonizing through the wetland, the water depth may be increased to approximately 25-40 cm and the flow rate may also be increased.

A uniform water depth is necessary to prevent future issues such as channeling of water, open water spots due to plants being unable to grow because of excessive water depth, or conversely, drying out. The vegetation selected for the CWTS (*C. aquatilis* and aquatic mosses) are known to form peat, which will accrete in a generally uniform manner through the CWTS as long as it is initially planted uniformly and bare spots are addressed through commissioning.

2.3.4. Flow Field

A uniform flow field is necessary in the CWTS to avoid channeling, as the plants will avoid higher flows and preferentially establish in low flow areas. As such, to ensure a uniform flow field, a distribution mechanism at the inflow and outflow areas of the CWTS is needed and will be addressed in the final wetland engineering design.

2.3.5. Water Velocity

Linear flow velocity for the CWTS was selected to be no greater than 35 cm/s according to the Hjulstrom Curve (Figure 7) based on the selected substrate of coarse sand (grain diameter = 1mm). This maximum linear flow velocity is considered to be highly conservative, as it is based on resuspension of substrate to avoid scouring, and does not account for plant roots retaining the sediments in high flows. However, references specific for maximum flow velocity in CWTS recommend the ideal flow velocity for treatment should be less than 15 cm/s for effective treatment capacity (ITRC, 2003). Therefore, the flow through the CWTS should ideally be less than 15 cm/s under normal conditions, and not exceed 35 cm/s under high flow events. These flow rates should be tested in a site-specific manner through the reclamation research program, as actual tolerances will vary depending on plant species, substrates used, and targeted treatment processes.

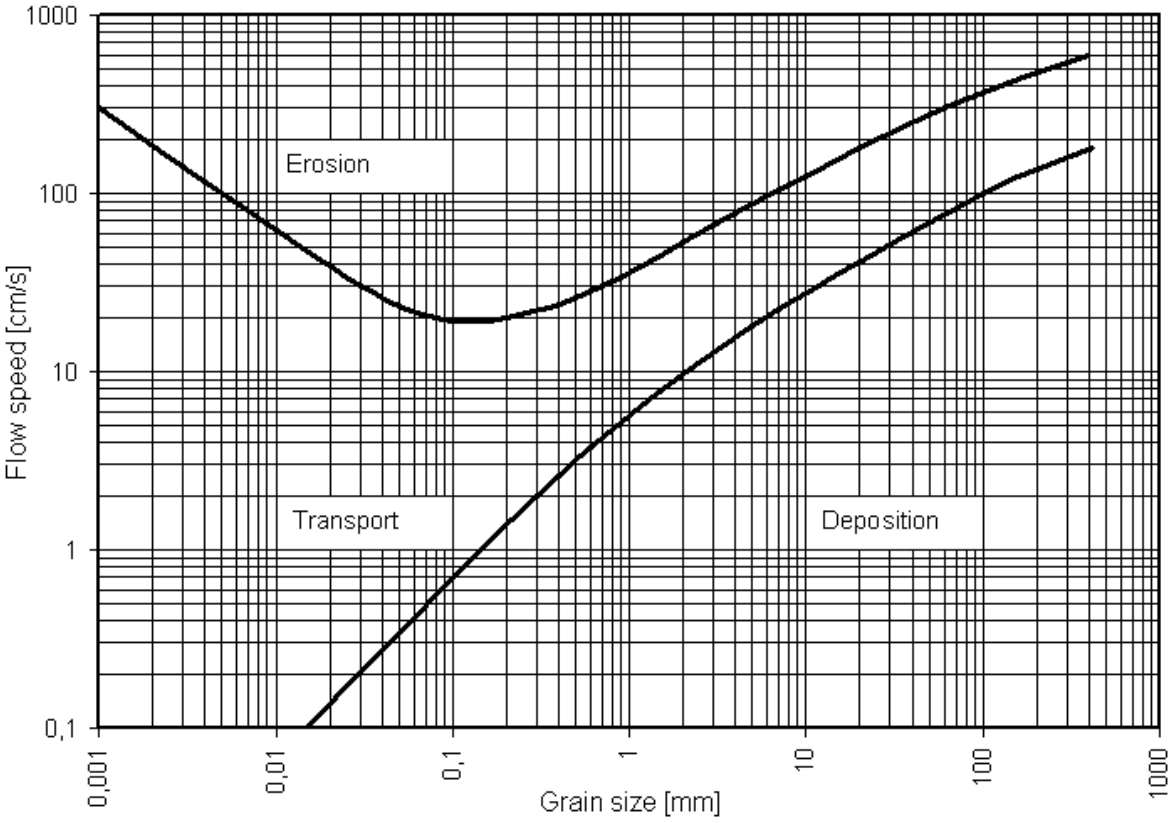


Figure 7 – Hjulstrom curve, demonstrating flow speed associated with resuspension of soils.

2.4. Removal Rate Coefficients and Calculations

An important factor for wetland design is the rate of treatment, also known as the treatment rate coefficient (*k*). The treatment rate coefficient is based on the treatability of a specific compound and the hydraulic retention time of the system, both of which are site-specific based on water chemistry, wetland designs, and characteristics of the system. The treatment rate coefficients in Table 5 are based on pilot-scale testing (Contango, November 2014) and are currently being refined with demonstration-scale (on site) testing (Contango March 2015, and Contango March 2016).

Based on pilot-scale testing for the Minto CWTS (Contango, November 2014), the treatment rate coefficient (k) applied for Se follows a zero-order reaction kinetic, while the rate coefficients for Cu and Cd follows first-order kinetics. In Equations 1-4, *Cf* is final concentration, *Ci* is initial concentration, V is volume of water in the system, and Q is flow rate. Using the removal rate coefficients (*k*) in Table 5 and Equations 1-4, parameters can be rearranged to solve for those of interest, such as the outflow concentration, given the available footprint in the WSP area.

Table 5 – Element treatment rate coefficient (*k*) values based on pilot-scale testing.

Element	*k* [1]
Cd	0.065
Cu	0.05
Se	0.0001

[1] All treatment rate coefficients are for first-order reaction kinetics except for selenium which is a zero-order reaction rate kinetic.

$$k = \frac{-\ln\left(\frac{Cf}{Ci}\right)}{V} \, x \, Q$$

Equation 1. Equation for calculation of first-order removal rate coefficient.

$$Cf = Ci \times e^{-k \, x \, \frac{V}{Q}}$$

Equation 2. Equation for calculation of first-order removal rate coefficient, rearranged to solve for outflow concentration.

$$k = \frac{(Ci - Cf)}{V} \, x \, Q$$

Equation 3. Equation for calculation of zero-order removal rate coefficient.

$$Cf = Ci - k \, x \, \frac{V}{Q}$$

Equation 4. Equation for calculation of zero-order removal rate coefficient, rearranged to solve for outflow concentration.

2.5. Thermodynamic minimum concentrations

Through all periods of pilot-scale testing, the outflow concentration of copper for any given system did not vary significantly, indicating that the thermodynamic minimum for copper removal has likely been approached within the system, with an average outflow concentration of 18 ± 5 µg/L. That is to say, making the wetland bigger using any of the tested designs would not result in a further significant decrease in outflow copper concentration, but rather, design alterations would need to be made in order to attempt to achieve lower outflow concentrations. It remains to be experimentally determined through the on-site demonstration-scale testing whether it is possible to achieve lower outflow concentrations of selenium and copper with this wetland design. Thermodynamic minimums developed through the site-specific testing for the Minto CWTS, have been applied to the conceptual performance modelling of a full-

scale CWTS (SRK, 2016a), these are: cadmium, 0.000009 mg/L; copper, 0.015 mg/L; and selenium, 0.004 mg/L. This is not to say that lower concentrations may not periodically occur through the CWTS treatment, however these are values that should be consistently attainable if sufficient retention time is provided, and lower concentrations and not expected to be consistently achieved.

2.6. Evapotranspiration

Evapotranspiration is another plant-mediated CWTS characteristic that must be considered (Beebe et al., 2014; Haakensen et al., 2015). Evapotranspiration can be defined as the total sum of water removed from a CWTS by evaporation from surface water of the system and transpiration from plant leaves. The degree of evapotranspiration is dependent primarily on the plant species/ecotype and controlled by meteorological conditions such as air temperature, relative humidity, and wind. Evapotranspiration drives an important aspect of the hydrology of a CWTS by enhancing the effective treatment area of surface flow wetlands.

Evapotranspiration can also affect the perceived treatment in a CWTS because removing water from the system concentrates the constituents, but also results in an increase of the hydraulic retention time (Beebe et al., 2014). That is to say, when evaporation and transpiration levels are high, an increased CWTS footprint may actually result in a higher outflow concentration of elements than is seen partway through the system, even though the total load of an element is lower at the outflow based on treatment.

Trials are being undertaken through the reclamation research program with the demonstration-scale CWTS to determine and model the effects of evapotranspiration on load removal, and current estimates are that approximately 1cm of water will be lost per day (equal to ~10 L per square meter of CWTS).

3. CWTS Implementation Overview

3.1. Timeline overview

The conceptual schedule for implementation of the CWTS is currently being refined through the reclamation research plan as part of the RCP. Through the reclamation research plan the monitoring and maintenance requirements for closure will be refined (research is currently planned through 2018 with the on-site demonstration-scale CWTS). Table 6 therefore provides the best estimate of site-specific maintenance and monitoring requirements for the CWTS based on information at this time.

Table 6 – Conceptual timeline.

Closure		CWTS Year	Period	Comments
Period	Year			
Interim and Active Closure	5	0	Construction	Construction and planting of system.
Post Closure I	6-7	1-2	Commissioning	Water from outflow can be recycled through CWTS or be sent to active water treatment if additional treatment is necessary.
	8-10	3-5	Early operation	System meeting performance targets, with frequent monitoring ongoing to refine site-specific long-term monitoring and maintenance requirements of CWTS. Minor adjustments may still be needed.
Post Closure II	11-20	6+	Long-term operation	Performance monitoring at reduced frequency, for confirmatory purposes. Maintenance as needed.

3.2. Construction

Physical construction of the CWTS is described in Appendix A (SRK, 2016b), and therefore only vegetation is covered here. Either before or after planting, a bed of straw and/or wood chips, approximately 5-10 cm deep, should be laid on top of the substrate to serve as an initial source of carbon for the beneficial microbes and to initiate reductive processes and contribute to accretion in early years while there is less plant matter being produced. If planted at the same density (5 plants per square meter) as the demonstration-scale CWTS (on site), *Carex aquatilis* would take approximately 2 years to fill in the CWTS, and based on results from the demonstration-scale CWTS, it is expected that a range of 3 – 5 *C. aquatilis* plants per square meter would be sufficient to fill the CWTS within 2 years. It is recommended

that the aquatic mosses be added much more densely than originally distributed in the demonstration-scale CWTS, in order to avoid needing to add additional moss through the year. Approximately 3,000 cm^3 of moss (moist but compressed) is recommended per m^2 of area within the CWTS. It is expected that a team of 4 people can plant a hectare of wetland per day. It is expected that a sufficient quantity of plants for the full-scale wetland can be sourced from the demonstration-scale CWTS, and the W10 area (the original source of plants for the demonstration-scale CWTS), if required additional borrow sources will be utilized.

3.3. Commissioning

The time period between construction of the CWTS and achieving expected treatment performance is referred to as the commissioning period. This period is needed for operational adjustments (e.g., raising water depth, modifying outflow patterns), and for plant and microbial populations to establish and mature. Based on pilot-scale testing (Contango, November 2014), and preliminary results from on-site demonstration-scale operations (Contango, March 2016), the estimated commissioning period for the full-scale system is 4 – 8 months (of active growing season, or 1 – 2 calendar years). Progress through the commissioning period is assessed through a monitoring program, such as the conceptual schedule inTable 7.

The commissioning period involves guiding the CWTS from construction through to early operation. Commissioning activities can be divided into the categories of: flow management, vegetation establishment, and establishment of conditions for treatment. Flow management includes ensuring the flow field is as uniform as reasonably possible, with wide inflow and outflow berms. Flow management is addressed in Appendix A (SRK, 2016b). Establishment of vegetation will also aid in flow management, and based on observations from the on-site demonstration-scale CWTS (Contango, March 2016), is expected to take a period of approximately 2 years depending on seasonal variations and planting densities. During this time, it may be necessary to supplement vegetation in some areas of the CWTS depending upon transplantation success. The on-site demonstration-scale CWTS had >95% survival of planted vegetation, and a similar rate would be expected at full scale. Variable water depths can sometimes aid in vegetation establishment, and slow (or stagnant) water flows can aid in development of microbial populations. Therefore, during the first months of commissioning, water may be intermittently pumped to the CWTS and allowed to evaporate (but not allowed to dry out), rather than a constant flow. With an estimation of ~1 cm evapotranspiration per day (with *C. aquatilis* vegetation), water could be flowed through the CWTS for 1 week of every month, with the remaining time non-flowing.

4. Monitoring

Monitoring and maintenance schedules contained herein are conceptual, and are being refined with the guidance of the reclamation research plan. The monitoring parameters and schedules are selected to inform potential maintenance needs, and in turn, possible maintenance activities are designed to address performance considerations. The conceptual schedules outlined here are conservative, but not prescriptive. Monitoring includes testing for performance (i.e., decrease of cadmium, copper, and selenium concentrations), and during earlier periods (commissioning, early operations Table 1, Table 2). Explanatory parameters are quantifiable aspects of a CWTS environment that can explain the performance of a CWTS. The end goal of explanatory parameters is to develop a decision management tree to guide maintenance activities associated with treatment performance of the CWTS. This allows for prediction of potential performance issues so they can be addressed before they occur, or the ability to detect and appropriately respond to upsets. There are many parameters that can be included as explanatory parameters, and it is important to identify those which are informative to the constituents being treated, and the site-specific context. For the Minto Mine CWTS, the explanatory parameters are being developed through a phased reclamation research plan, and pilot-scale testing has thus far identified pH, dissolved oxygen (DO), oxidation-reduction potential (ORP), conductivity, temperature, total organic carbon (TOC), nitrate, and soil redox potential as useful to understanding the performance of the CWTS. The explanatory parameters are continuing to be refined through the on-site demonstration-scale operation. Actual parameters and frequency of testing should be guided by the scientific analysis of the results from the previous operational period, therefore, the commissioning will be guided by the on-site demonstration, the early operations by the commissioning, and so forth. Maintenance of the system should be performed as necessary, and based on the results of monitoring activities.

4.1. Monitoring during commissioning (Years 1 – 2)

The commissioning period involves the most rigorous monitoring, to ensure the system is brought online effectively. Because biological systems are dynamic and could take several paths to maturation, rigorous monitoring during the commissioning phase is important to track progress and make corrections as necessary, allowing for the system to be guided through the desired maturation path. During the commissioning period, both performance and explanatory parameters are monitored in order to assess both the effectiveness and mechanisms of function of the CWTS. Some of these parameters would conceptually require weekly monitoring (e.g., visual inspections, water quality testing for total and dissolved metals, nitrate, pH, dissolved oxygen, oxidation reduction potential, and monitoring of plant health and establishment), while others would be tested less frequently. For example, water may be tested monthly for dissolved organic carbon and substrates for redox potential, while parameters such as substrate microbiology, nutrient and organic carbon content may be tested seasonally, and soils metals assays may occur on an annual basis. Parameter lists and testing schedules are being refined through the on-site demonstration-scale CWTS as part of the reclamation research plan.

Table 7 – Conceptual monitoring schedule.

Closure Period		Year 1 - 5 (Interim and Active Closure)	Year 6 - 10 (Post Closure I)								Year 11 -20 (Post Closure II)			
CWTS Stage		Construction	Commissioning (Year 6-7)				Early Operation (Year 8-10)				Long-term Operation (Year 11-20)			
Site	Description	No Monitoring Required	Water	Substrate	Flow	Plants	Water	Substrate	Flow	Plants	Water	Substrate	Flow	Plants
CWTS IN	CWTS Inflow	NA	W,M	NA	W,M	NA	M	NA	M	NA	SSF	NA	SSF	NA
CWTS MID	Within CWTS	NA	W,M	M,SSF,A	W,M	W,M,A	M	M,A	NA	SSF,A	A	A	A	A
CWTS OUT	CWTS Outflow	NA	W,M	NA	W,M	NA	M	NA	M	NA	SSF	NA	SSF	NA

W – weekly; M – monthly; SSF – spring, summer, fall; A – annually; NA – not applicable.
Where two frequencies are separated by a comma, multiple parameters are being tested, and on different schedules.

4.2. Monitoring during early operation (Years 3 – 5)

The CWTS will be transitioned from the commissioning monitoring schedule and considered to be operating in "Early Operations" once the explanatory parameter of targeted soil redox and outflow performance objectives for copper and cadmium are met. At this period targeted performance for selenium may or may not be met, depending on the depletion rate of residual nitrate concentrations. During Early Operations period, the CWTS is expected to be consistently meeting performance targets; however, frequent monitoring is continued to refine site-specific long-term monitoring and maintenance requirements of this system (Table 3). During early operations, the CWTS should be meeting performance targets, but with monitoring of performance and explanatory parameters ongoing to refine site-specific long-term monitoring and maintenance requirements of CWTS. Minor operational adjustments may still be needed during this period. Parameters that would conceptually continue on a monthly monitoring schedule include visual inspections, water quality testing for total and dissolved metals, nitrate, pH, dissolved oxygen, oxidation reduction potential, substrate redox potential, and general plant health. Parameters such as substrate microbiology, nutrient and organic carbon content, and metals assays may occur on an annual basis. Parameter lists and testing schedules are being refined through the on-site demonstration-scale CWTS as part of the reclamation research plan.

4.3. Monitoring during long-term operation (Years 6 – 10)

During long-term operation of the CWTS, monitoring is conducted mostly for performance, with a reduced frequency of monitoring for explanatory parameters (as defined from commissioning and early operations periods). The decision tree that was developed during demonstration-scale operation (and refined in full-scale commissioning and early operations) is applied to guide maintenance activities. During long-term operations, the CWTS should be meeting performance targets, with monitoring of performance and only those key explanatory parameters defined as leading indicators of performance (through the reclamation research plan and early operations of the full-scale CWTS). Conceptually, seasonal testing will be performed for performance (dissolved metals), and soil redox. Parameter lists and testing schedules are being refined through the on-site demonstration-scale CWTS as part of the reclamation research plan, and will be further refined through testing in commissioning and early operations of the full-scale CWTS.

5. Maintenance and operational adjustments

Maintenance of engineering and flow-management structures associated with the CWTS are covered in Appendix A (SRK, 2016b), and only performance related aspects are addressed here. Activities commonly associated with maintaining the performance of a CWTS are detailed in Table 8. The frequency of these activities should be guided by the monitoring activities, and is being refined through the on-site demonstration-scale CWTS as part of the reclamation research plan. A decision matrix based on the explanatory parameters is being developed through the reclamation research plan, and will be used to guide the operation of the full-scale CWTS. Additionally, periods of more intensive monitoring during commissioning and early operations will further guide the appropriate frequency of site-specific maintenance and refine the decision matrix. During early operations, minor adjustments may still be needed to the system, such as addition of solid phase organic materials (e.g., straw or wood chips), and corrections to non-uniform flows (e.g., adding sand bags at inflow and outflow of cells).

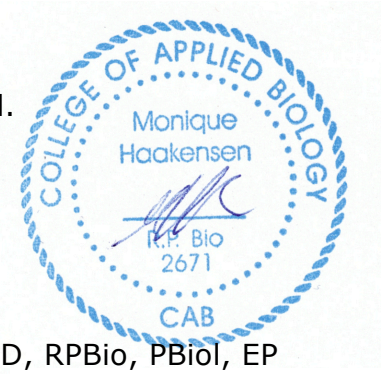
Table 8 – Conceptual maintenance actions.

Observation	Monitoring	Item Addressed	Action	Follow up
Soil redox drifting upwards	Nutrients, TOC, microbiology	Decrease of treatment effectiveness due to low microbial activity or lack of food for microbes.	Add nutrients (fertilizer), and/or organic carbon.	Soil redox testing weekly until within stable range. Microbiology, TOC and/or nutrients testing bi-weekly until within desired range.
	Flow rates	Water volume to carbon ratio is too high.	Check inflow mechanism.	If ongoing issue (and not addressed by adding organic carbon), adjust inflow mechanism.
Water depth decreasing	Flow rates, accretion	System is designed to accrete (increase mass of soil), which causes soils to build within CWTS. Expected rate of 0.5-2mm/year (Robinson and Moore, 1999).	Increase height of outflow (and inflow) berms.	Record accretion rate to schedule for next depth increase event.
Undesirable plants	Vegetation monitoring	Channeling or decreased treatment effectiveness due to competing microbial habitats.	Remove undesirable plants.	Monitor to ensure undesirable plants have successfully been removed.
Wildlife interferences	Visual monitoring	Non-uniform flow field due to wildlife interferences (e.g., beaver dam) and decreased treatment performance.	Remove wildlife interferences.	Monitor to ensure interference is not re-established.
Channeling	Flow rates	Channeling can be due to undesirable plants, non-uniform flow fields, or flow rates beyond CWTS design.	Check inflow mechanism.	If ongoing issue, adjust inflow mechanism.
Flow blockages	Flow rates, visual monitoring	Blockages can be due to debris (fallen branches) and wildlife activity (beavers).	Remove blockages.	Adjust monitoring frequency accordingly.
Poor plant health	Nutrients	Decreased treatment effectiveness due to less active treatment zone if plants not actively transpiring and moving water through root zone.	Add nutrients (fertilizer).	Nutrients testing bi-weekly until within desired range.

6. Closure

We trust the information herein satisfies your present requirements. Should you have any questions, please contact the persons listed below. We appreciate the opportunity to provide the services detailed in this report, and look forward to discussing any comments you may have.

Regards,
Contango Strategies Ltd.



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Minto Creek Constructed Wetland: Physical Infrastructure Preliminary Design

Prepared for

Minto Explorations Ltd.



Prepared by

SRK Consulting (Canada) Inc.
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Minto Creek Constructed Wetland: Physical Infrastructure Preliminary Design

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1 Introduction

Minto Explorations Ltd. (Minto) asked SRK develop designs for the necessary physical infrastructure to support a Constructed Wetland Treatment System (CWTS) proposed to be constructed within the footprint currently occupied by the Water Storage Pond (WSP) as part of mine closure works at Minto Mine, central Yukon Territory. This report describes the design basis and the accompanying drawings provide the preliminary design details.

This physical infrastructure preliminary design report was developed to support Contango Strategies' design of the treatment wetland (Minto Mine Reclamation and Closure Plan – Preliminary Design Report for Treatment Wetland (Contango 2016)). Details related to wetland establishment, monitoring, and maintenance are provided in Contango's design report, along with a thorough discussion of treatment mechanisms, on-site wetland trials (completed, in-progress, and planned), and performance expectations.

2 Background

Water Licence QZ14-031 was issued to Minto in 2015. The licence included a condition requiring Minto to submit a revised reclamation and closure plan (RCP), and the submission date for the revised RCP was extended to August 5, 2016 via a December 2015 amendment to QZ14-031. Minto's 2016 Reclamation and Closure Plan will be the sixth major revision of the site closure plan, and will be referred to as 'RCP 2016-01' in this document.

The CWTS preliminary infrastructure design is one component of RCP 2016-01.

2.1 Current Layout of Site

The Water Storage Dam was constructed during initial mine construction in 2006. It was constructed across the Minto Creek valley at a narrow point near the eastern limit of Minto's quartz claims and represents the eastern-most major infrastructure facility at Minto Mine. The dam construction formed the WSP reservoir, which was initially used for mill water supply and later as a clean water reservoir for storing water prior to discharge to Minto Creek. The outline of the WSP and the Water Storage Dam are shown in Drawing WC-01.

2.2 Closure Phases

For the closure and post closure periods, planning and management of site activities are guided by the Minto RCP. This includes the identification of distinct closure phases with defined activities and performance expectations:

1. Active Closure;
2. Post Closure I (transition to passive treatment); and
3. Post Closure II (passive treatment only).

3 Design

3.1 Overview

During Active Closure, the Water Storage Dam will be decommissioned and the footprint of the WSP will receive fill and be graded to allow the subsequent construction of a Constructed Wetland Treatment System (CWTS), which is also referred to as the Minto Creek Constructed Wetland. The CWTS will receive water from E3 Conveyance Channel in the closure conveyance system, which receives surface water from nearly all of the disturbed areas of the mine site that lie within the Minto Creek catchment. Flows up to 3000 m^3/day[1] will route through the CWTS, with inflows controlled by inlet control structures. To protect the CWTS from high flows, the inlet control structures are designed to convey higher flows through a spillway system to discharge into a high flow bypass channel. Details are presented in Drawings CW-03 in Appendix A.

During Post-Closure 1, commissioning of the CWTS will take place and CWTS effluent will be managed by a combination of reservoir and pumping-and-piping system, with water that meets discharge standards discharged off site and water that does not meet discharge standards pumped to one of the open pits for storage. As discussed in the Section 7.7.1 of the RCP the water treatment plant used during operations will remain in operational until WUL criteria can be achieved consistently without its utilization. Transition from Post-closure 1 to Post-closure 2 will occur when the CWTS effluent adequately meets performance objectives, at which time the effluent reservoir will be connected to the high flow bypass channel and the system will transition to being passively managed

Drawings WC-01 through WC-04 provide the preliminary designs for the physical infrastructure of the CWTS and bring in relevant conveyance design details from other components of the Minto Mine RCP for reference, including details from Appendix G2, *Closure Water Conveyance System Design Update Report, Minto Mine, Yukon* (SRK 2016a).

The following sections generally are organized sequentially by order in which the various components will be constructed.

3.2 Design Objectives, Considerations and Criteria

Summarized in this section are the objectives, considerations, and criteria for the preliminary CWTS design. Physical design elements are illustrated in the appended preliminary drawing package. Design criteria that are critical to the biological efficiency of the wetland system to sequester soluble metals were developed by Contango. These design elements are included in the physical wetland design and the water conveyance dynamics.

3.2.1 Head Pond and Wetland Intake Structure:

- Design flow rate for the passive treatment wetland system is based on a typical average monthly summer flow experienced in June, July and August. The maximum targeted flow rate

[1] Range of flows received by CWTS is expected to range from zero (during summer dry periods) to a maximum of 3000 m^3/day (under extreme high runoff conditions).

to be diverted into the wetland system is 3000 m^3/day or approximately 0.035 m^3/s. 3000 m^3/day is the upper limit of the flow rate that the CWTS is intended to handle.

3.2.2 Wetland Cell and Conveyance Channel Geometry:

- A 3:1 aspect ratio (Length to Width) is used to site the location of the passive treatment wetland cells. The cells are typically 15 m wide and 45 m long. Due to the limited width of the valley, space was limited in some areas- the cells were designed as typical 15 m wide and in some cases it was necessary to shorten the length of the cells to fit the site configuration.
- Substrate depth within the cells is designed to be a minimum of 0.5 m, substrate depths are illustrated in the design drawings to be 1 m thick and submerged by 0.3 m for a total of 1.3 m of initial design height above liner to the water surface in each cell. The substrate material will consist of sand with 2 to 7% by volume as organic matter (e.g. wood chips, straw).

3.2.3 Seepage Management:

- Seepage losses from the wetland cells and groundwater ingress into the cells is managed by inclusion of a low hydraulic conductivity geomembrane liner. The liner will prevent both groundwater from entering and seepage from leaving the CWTS and will avoid the performance uncertainty that would accompany unquantified inputs and losses of both flow and geochemical load.

3.2.4 Erosion Management:

- Internal erosion protection of the wetland pond system is limited to rip-rap channel protection because the CWTS inlet control structure is to be designed to limit inflows to a maximum flow rate of 3000 m^3/day.
- Distinct from limits on daily volumes, considerations around wetland plant establishment and propagation patterns require limits on water velocity as follows: maximum velocity 0.35 m/s, and typical velocities of <0.15m/s (Contango 2016).

3.2.5 Restrict Dam Classification for Designed Infrastructure:

- Design intentions have been focused on precluding the classification of any design infrastructure as a dam as defined by the Canadian Dam Safety Guidelines (CDA 2007) and dam classification guidelines in the Yukon Territory. CDA defines a dam as a constructed barrier that is at least 2.5 m tall and retains at least 30,000 m^3 of water. The Yukon Territory's definition of a constructed dam without a licence for a Schedule 7 Quartz Mining Undertaking as having a height greater than 3 m and less than 10,000 m^3 of water storage (Yukon Government 2012).

3.2.6 Exclusions

The following elements are outside of the limits of the preliminary physical infrastructure design:

- Establishment and success of wetland vegetation in the CWTS,
- Treatment performance of the CWTS,

- Revegetation of disturbed surfaces associated with the construction of the CWTS, and
- Construction compaction specifications of fill materials are not specified for this preliminary level of design. An assessment of available material properties is required at future stages of design to determine compaction requirements.

3.3 Constructability and Sequencing

The overall wetland layout and the configuration of the cells and intake structures have considered a long-term closure of the Minto site and the future needs of this site beyond the Post Closure 1 timeline. Intake and conveyance structure designs to the wetland ponds have been considered for their ability to be adaptive (i.e. allowance for the modification of conveyance elevations and flow rates to allow vegetation to establish).

In future stages of closure design it is recommended that a staged construction and decommissioning plan be developed to optimize water management resource needs and material handling. Additionally, a phased commissioning plan for a constructed closure conveyance system and wetlands must be developed.

Decommissioning of the Water Storage Pond Dam is required to construct the wetland and diversion channels. Initial decommissioning efforts require that the reservoir be drawn down and incoming flows routed around the construction area to allow for the partial excavation of the dam. Flow routing is expected to include excavating a water retention sump and back pumping water to the Main Pit to manage sedimentation and through pumping water from the Main Pit to Minto Creek. Retention sump locations details are provided in Drawing CW-02 in Appendix A.

Final decommissioning of the WSP dam will include the construction of the High Flow Bypass Channel that will convey flow around the wetlands. Additionally, it will require the removal of the sump and pumping diversion systems to allow flows to enter the wetland pond system. The infrastructure design, alignments, and specifications are located in the drawings in Appendix A.

Access for post-closure monitoring and maintenance at the CWTS is to be maintained after construction. The pond system is designed with a wide berm access of 4 m around the perimeters of each wetland cell to accommodate a vehicle or small excavator.

3.4 Foundation Preparation and Base Construction

To construct the passive treatment wetlands and provide a graded profile to the wetland area the existing WSP Dam will be decommissioned. Decommissioning of the WSP dam is illustrated in Drawing CW-02 located in Appendix A. The Dam is constructed of coarse waste rock and fine grained materials as segregated zones. During the decommissioning of the dam the segregated materials are repurposed in the construction of the wetland system. Coarse rock will be used to build up the basal area of the pond system, fine grained materials are to be used as separation layering between geomembranes and coarse materials.

During the decommissioning of the WSP Dam an existing access road will potentially require upgrades for haul truck access within the former ponded area. Material stockpiles, load and haul routes within the wetland area should be constructed as dam decommissioning advances.

3.5 Wetland Intake Flow Control Structure

A Primary Head Pond is designed to buffer turbulent flows exiting the E3 conveyance channel stilling basin. The pond elevation (715 masl) will be controlled by two hydraulic control structures: High Flow Discharge Channel and the Wetland Intake Structure. The Secondary Head Pond Intake Structure will convey 1% of the total flow entering the Primary Head Pond into the Secondary Head Pond the remainder of flows enter the High Flow Bypass Channel. Water entering the Secondary Head Pond will then be moderated where the design flow for the wetland system is conveyed in the wetland intake structure and the remainder of flow is conveyed through a spillway into the High Flow Bypass Channel.

The Secondary Head Pond and Wetland Intake structure are designed to reduce the intake flows and convey the expected design flows from the Primary Head Pond to the wetland cells. Flows exceeding this rate will be diverted to the High Flow Bypass Channel. Discharge flow from the wetland system is discharged to the High Flow Bypass Channel down gradient of the wetland system. Surface water quality monitoring station W50 is located approximately 50 m downstream of the confluence of these two flows.

At this stage of design the wetland intake structure has not been designed in detail. Conceptually, hydraulic principles that govern open channel flow have been considered to confirm that the system of ponds, inlet structures, and high flow spillways can reduce peak flows adequately to limit CWTS inflows to the design peak flow rate. The wetland infrastructure design is capable of separating a maximum of 3000 m^3/day from the mine closure conveyance system and conveying the flow through the wetland cells. It should be noted that this design peak flow rate is based on treatment performance limitations, and not on limitations of the hydraulic system to pass substantially greater flow volumes.

Future stages of design should consider an options analysis for common hydraulic structures such as weirs, flumes, and orifice flow control structures. Options to consider include orienting the inlet structure to be south facing to maximize early spring thaw, and applying a coating to the inlet structure to maximize solar heat absorption to melt and shed ice buildup.

3.6 Passive Treatment Wetland Cells and Conveyance Channels

Water depth within the wetland cells is designed to a targeted 0.15-0.4 m depth. After vegetation planting and during the initial growing seasons the water level in the cells will be controlled to be 0.2 m. Once the vegetation is rooted and acclimatized the wetland cells, water depth will be increased to 0.3-0.4 m. Section and plan details illustrating the CWTS designs are provided in Drawing CW-02 and CW-04 in Appendix A.

Internal cell-to-cell conveyance channels are designed as an excavated 4 m wide swale with 2:1 side slopes and a 1% slope. The mild channel slope is designed to have a 350 mm layer of rock

rip-rap with an average diameter of 200 mm. Flow velocities through the cell spillway sections are measured as 0.3 m/s and a water depth of 30 mm.

Underlying the passive treatment wetland cells and the cell-to-cell conveyance channels is a geomembrane to manage seepage loss and ingress. It is recommended that all exposed geomembranes be covered with a minimum of 0.3 m of fine grained material to reduce its exposure to radiation from the sun.

4 Physical Monitoring and Maintenance

The construction, commissioning, and operation of the passive treatment wetland is expected to have a variable management requirements for the physical monitoring and maintenance. This section discusses what physical attention is needed for the wetland infrastructure. Effective implementation of the *Closure Adaptive Management Plan* (SRK 2016b) will mitigate conditions beyond those expected during closure and post closure care and maintenance programs.

During construction and dam decommissioning water within the footprint of the CWTS construction will require ongoing and active monitoring and maintenance. The commissioning phase will require a site presence to monitor flow rates at the wetland intake structure to confirm that the wetland system is receiving the design flow rates. During this phase it may be required to make alterations to the flow regulating structure. After the commissioning phase it is expected that the frequency of inspection and maintenance decrease over time. However, long-term physical monitoring and maintenance will be required for this wetland system.

Monitoring frequency and maintenance requirements should be re-assessed and adapted based on the judgement of a qualified person as the site transitions through Active Closure, Post-Closure I and Post-Closure II periods. The proposed schedule for the monitoring programs is provided in Section 7.12 of the RCP. The following subsections suggest some of basic monitoring and maintenance needs for the structure of the wetland system.

4.1 Active Closure

Physical monitoring and maintenance requirements during the construction period include:

- Routine water quality sampling for sediment and metal composition at the intake and discharge of the pond system.
- Assessing the grading of the channels between ponds to assess the flow paths. Flow through the channels should be unobstructed and non-eroding.
- During initial rooting stages of pond vegetation commissioning the water depth in the ponds are expected to be lower than the final operational elevation. Once advised that the rooting is sufficient the elevation in the ponds can be increased to the final elevation.
- Routine removal of vegetation and debris that may block channels.

- Assessment of initial signs of settlement and erosion of surfaces adjacent to the wetland system. If settlement within the conveyance network is observed then remedial construction activities should be initiated as described in the closure adaptive management plan.
- Ground water seepage assessments of the surrounding area to determine if there will be exposed seeps in the former WSP area that may affect the pond system. If seepage volumes are considerable and are leading to erosion of soils then remedial construction activities should be initiated as described in the closure adaptive management plan.

4.2 Post-Closure I

The closure monitoring and maintenance requirements will be developed at the end of the active construction phase as the site transitions to passive treatment. Typically the monitoring will assess the following geotechnical and hydraulic conditions:

- Measure the flow rates into the wetland system and determine if the flow rates are appropriate and if there are meeting objectives. Field samples should be collected to determine the water quality at the intake and discharge of the pond system.
- Assess the flow profile through and in between each wetland cell to document the development or extent of erosion or blockages.
- The intake flow structures should be assessed for operation and integrity.
- Surface erosion and settlement deformations.

4.3 Post-Closure II

An assessment of the existing state of the wetland system is required to identify the needs of monitoring and maintenance as the site transitions to the Post Closure II phase.

5 Recommendations

This section summarizes design requirement to advance the design of the wetland system from preliminary to a feasibility level.

- Limnology study of sediments in WSP to quantify chemistry and thickness to determine if additional design and management allowance is needed for the wetland pond system.
- Geotechnical investigation of the wetland foundation area and the High Flow Bypass Channel for the existence of permafrost and other geotechnical considerations which may have a potential for thaw consolidation and settlement.
- Site wide definition of material sources including defining dam materials and potential opportunity to use local soils and local quarries near the existing WSP Dam. The site-wide materials assessment should define the volumes and material types that are needed for the wetland construction.

- Evaluation of merit of routing the High Flow Bypass Channel through a purpose built rock quarry located in south abutment of dam.
- Assessment of the suitability of rip-rap in the current WSP dam spillway for use as riprap in the long term. Evaluation of the existing spillway rip-rap covering is needed to determine if the condition of the armouring is suitable for reuse for closure activities as foundation fill material for the wetland area.
- Excavation optimization for the materials removed from the WSP Dam which can be reused as foundation fill material for the wetland area.
- Hydrologic assessment of the transition of the High Flow Bypass Channel from the former WSP Dam area downstream to the Minto Creek basin.
- Evaluation and design of the hydraulic intake structure that moderated flow from the head ponds into the wetland system at the specified rate of 3000 m^3/day or the rate required during the final engineering and design stages.
- Reassessment of the need for a geomembrane to line the ponds and conveyance channels between ponds.
- Future assessment to the efficiency of the wetland pond dynamics may redefine the cell geometry needs and peak flow rate requirements defined by Contango.
- Head Pond: design requires further optimization such that the final design (excessive storage water volume or height of impounded water) does not become defined as a dam.
- Tailings Diversion Ditch: energy dissipation area requires additional design consideration to transition flow exiting the Tailings Diversion Ditch into the High Flow Bypass Channel. Additional design should focus on managing flow hydraulics and erosion.

This report, Preliminary Design: Physical Infrastructure for the Water Storage Pond Constructed Wetland Treatment SystemPreliminary Design: Physical Infrastructure for the Water Storage Pond Constructed Wetland Treatment System, was prepared by

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All data used as source material plus the text, tables, figures, and attachments of this document have been reviewed and prepared in accordance with generally accepted professional engineering and environmental practices.

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The opinions expressed in this report have been based on the information available to SRK at the time of preparation. SRK has exercised all due care in reviewing information supplied by others for use on this project. Whilst SRK has compared key supplied data with expected values, the accuracy of the results and conclusions from the review are entirely reliant on the accuracy and completeness of the supplied data. SRK does not accept responsibility for any errors or omissions in the supplied information, except to the extent that SRK was hired to verify the data.

6 References

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Appendix A: Drawings

Engineering Drawings for the Minto Creek Constructed Wetlands Minto Mine, Yukon, Canada

ACTIVE DRAWING STATUS

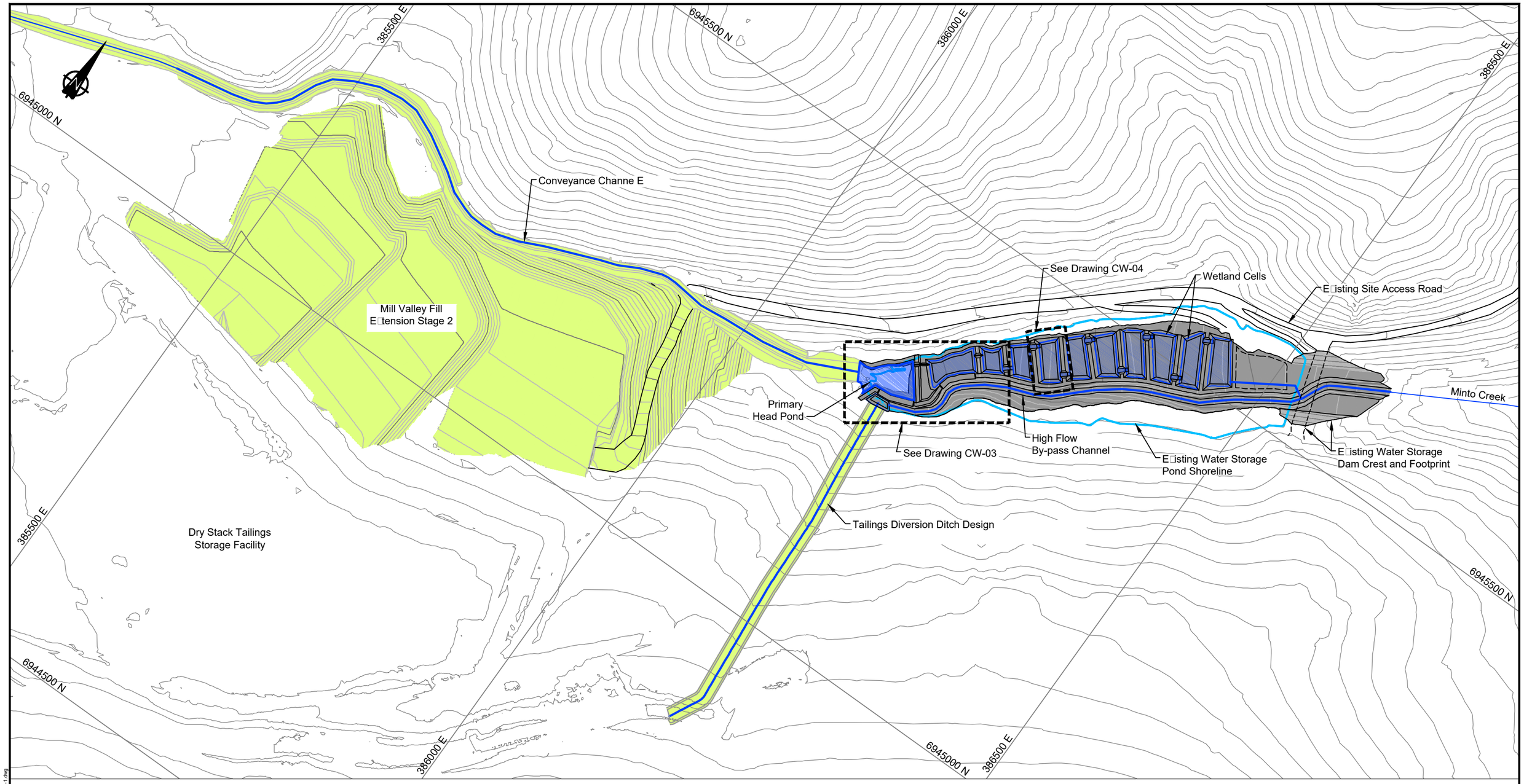
DWG NUMBER	DRAWING TITLE	REVISION	DATE	STATUS
CW-00	Engineering Drawings for the WSP Constructed Wetlands, Minto Mine, Yukon, Canada	A	July 29, 2016	Issued For Review
CW-01	General Arrangement	A	July 29, 2016	Issued For Review
CW-02	Wetland Cell Configuration	A	July 29, 2016	Issued For Review
CW-03	Primary Head Pond Area Layout	A	July 29, 2016	Issued For Review
CW-04	Typical Cell and Cell Spillway Layout	A	July 29, 2016	Issued For Review

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Revision A - Issued for Review
July 29, 2016
Drawing CC-00



NOTES

1. Contour Interval 1m typical though drawing package.
2. Coordinate system is in NAD83 UTM □one 8
3. Digital topographic surface has been provided by Minto E□ploration ltd.
4. Notes on this drawing apply to all other drawings within this drawing package unless otherwise advised.

LEGEND

- Constructed Wetlands Design
- Designed Infrastructure □Not Constructed□
- Design Toes □ Crests
- Flow Direction

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0 40 80 120 160 200

Scale in Metres

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REFERENCE DRAWINGS

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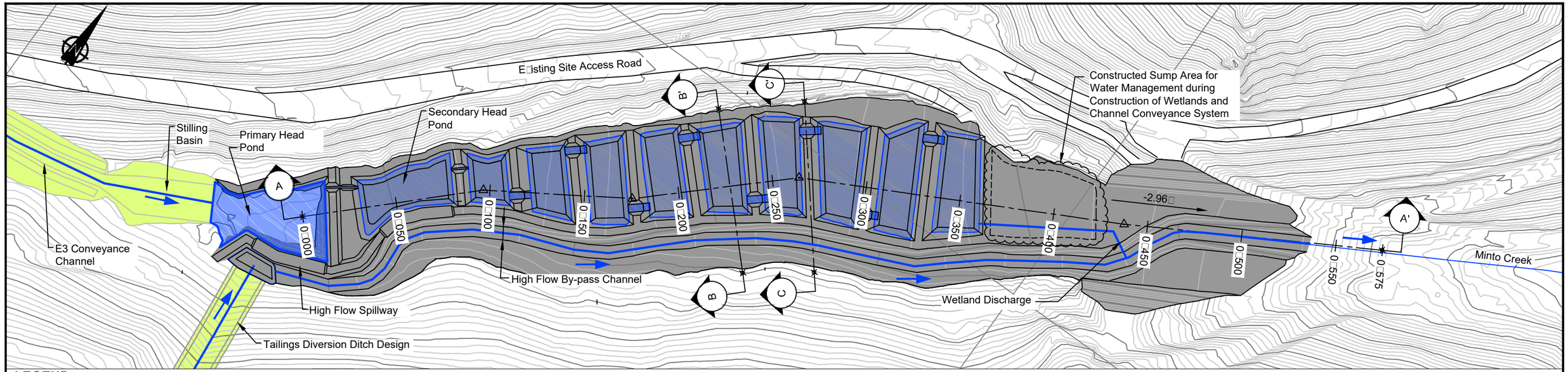
SRK JOB NO.: 1CM002.042

Preliminary Infrastructure Minto Creek Constructed Wetlands

DRAWING TITLE: General Arrangement

DRAWING NO.	SHEET	REVISION NO.
CW-01	2 OF 5	A

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LEGEND

- Wetland Foundation Fill Material
- Rip Rap
- Constructed Wetlands Design
- Designed Infrastructure (Not Constructed)
- Geomembrane
- Flow Direction

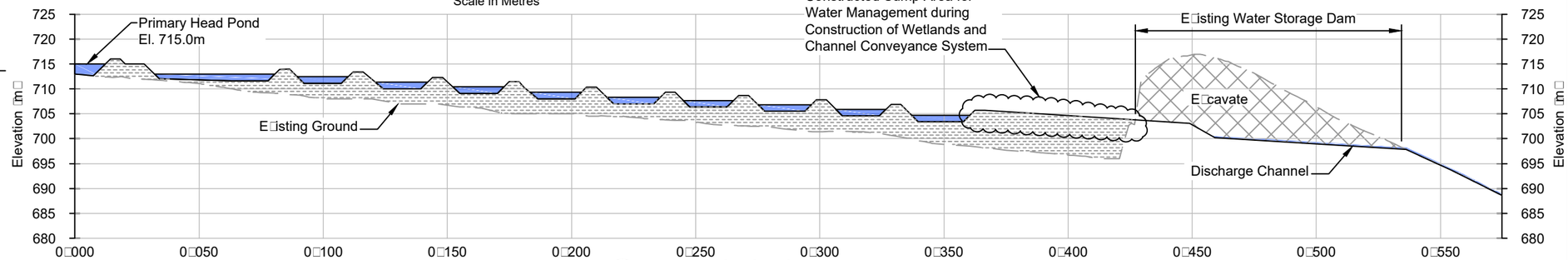
NOTES

1. During the decommissioning of the dam the segregated materials are repurposed in the construction of the wetland system
2. Road access within WSP footprint will be required for construction.
3. Material stockpiles, load and haul routes within the wetland area should be constructed as dam decommissioning advances.
4. Decommissioning of the WSP dam will include the construction of the High Flow By-pass Channel that will convey flow around the wetlands as per SRK Closure Conveyance Channel Design 2016.
5. Supporting features: E3 Stilling Basin, Energy dissipation pond, and the High Flow By-pass Channel are designed by SRK in the Closure Conveyance Design Report 2016.
6. Notes on this drawing apply to all other drawings within this drawing package unless otherwise advised.

CONSTRUCTED WETLANDS PLAN VIEW

0 20 40 60 80 100

Scale in Metres

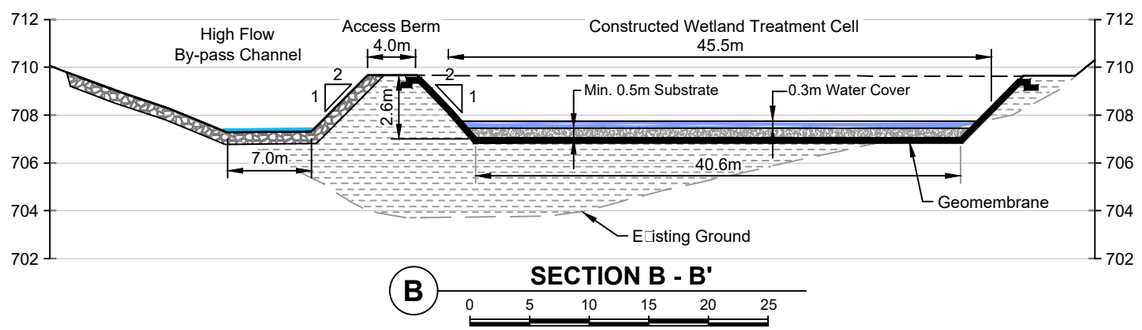


SECTION A - A'

A

0 20 40 60 80 100

Horizontal Scale in Metres
Vertical Exaggeration 2x

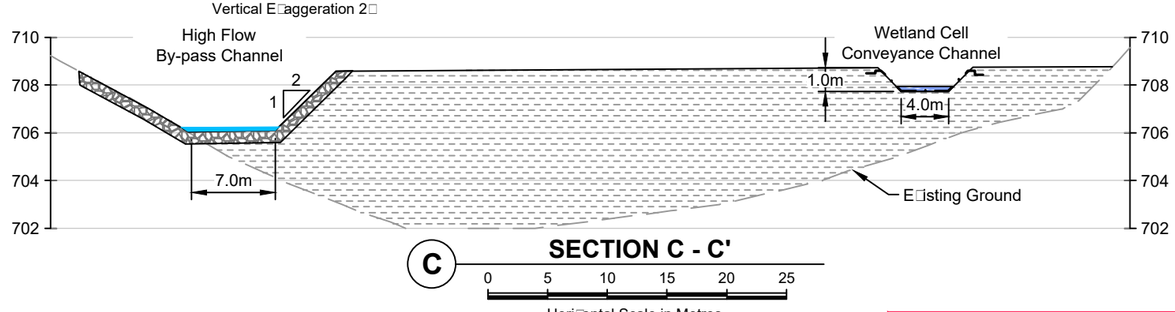


SECTION B - B'

B

0 5 10 15 20 25

Horizontal Scale in Metres
Vertical Exaggeration 2x



SECTION C - C'

C

0 5 10 15 20 25

Horizontal Scale in Metres
Vertical Exaggeration 2x

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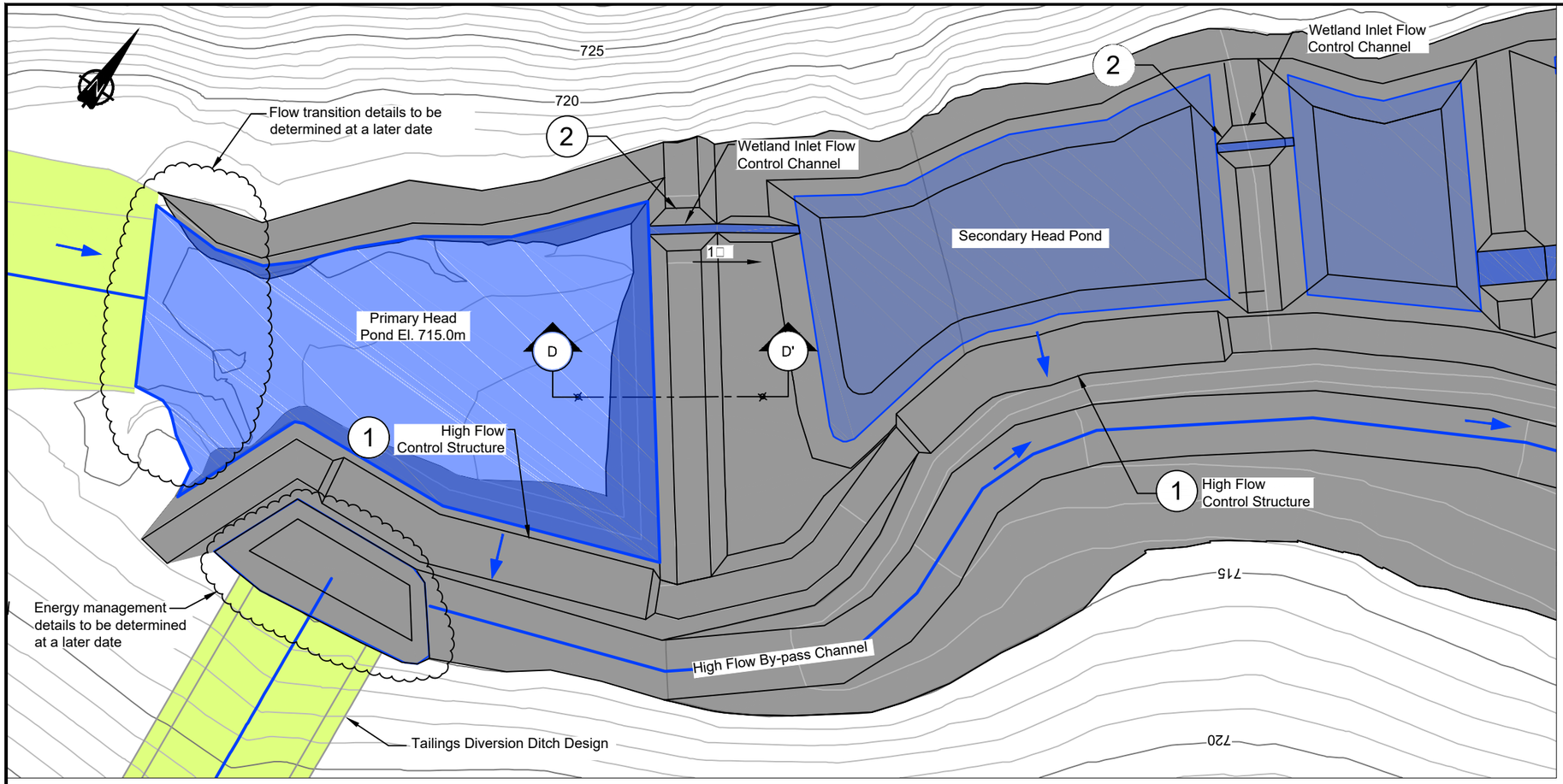
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Preliminary Infrastructure Minto Creek Constructed Wetlands

DRAWING TITLE: Wetland Cell Configuration

DRAWING NO.	SHEET	REVISION NO.
CW-02	3 OF 5	A

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LEGEND

- Wetland Foundation Fill Material
- Rip Rap
- Constructed Wetlands Design
- Designed Infrastructure (Not Constructed)
- Geomembrane
- Flow Direction

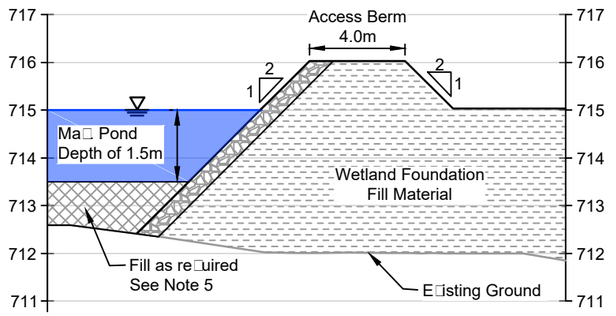
NOTES

1. The Wetland Intake structure is designed to convey the design flow rate from the Primary Head Pond to the Secondary Head Pond. Flows exceeding this rate enter the high-flow bypass channel.
2. Wetland intake structure to be designed at later stages of design to convey 3000 m3/day into the wetland.
3. High Flow By-Pass flow control structure to be designed at later stages of design.
4. High Flow By-Pass Channel and Stilling Basin designed by SRK Closure Conveyance Channel Design Report 2016.
5. Pond to be filled to avoid structures classified as dams

PRIMARY AND SECONDARY HEAD POND PLAN VIEW

0 10 20 30

Scale in Metres

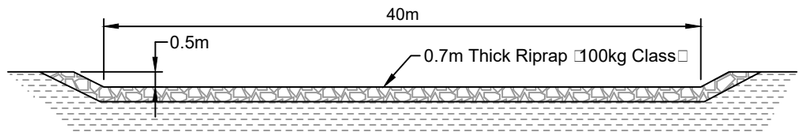


SECTION D - D'

D

0 2 4 6 8 10

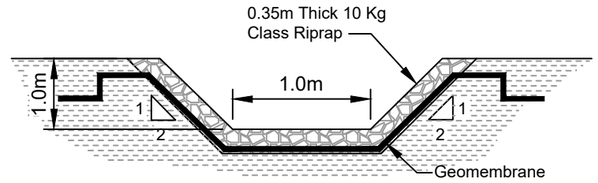
Horizontal Scale in Metres
Vertical Exaggeration 2X



DETAIL 1 - TYPICAL HIGH FLOW CONTROL STRUCTURE

1

NTS



DETAIL 2 - TYPICAL WETLAND INLET FLOW CONTROL CHANNEL

2

NTS

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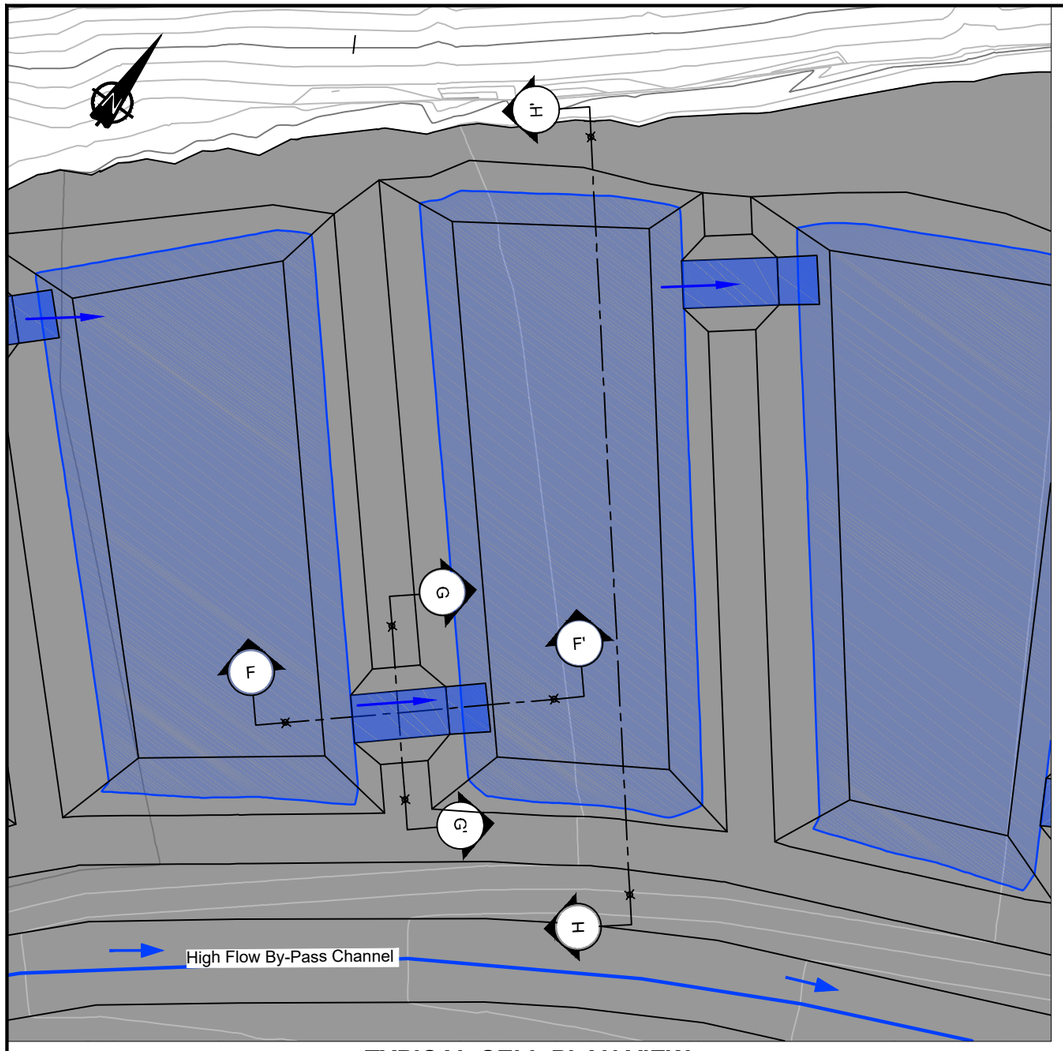
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Preliminary Infrastructure Minto Creek Constructed Wetlands

DRAWING TITLE: Head Pond Area Layout

DRAWING NO.	SHEET	REVISION NO.
CW-03	4 OF 5	A

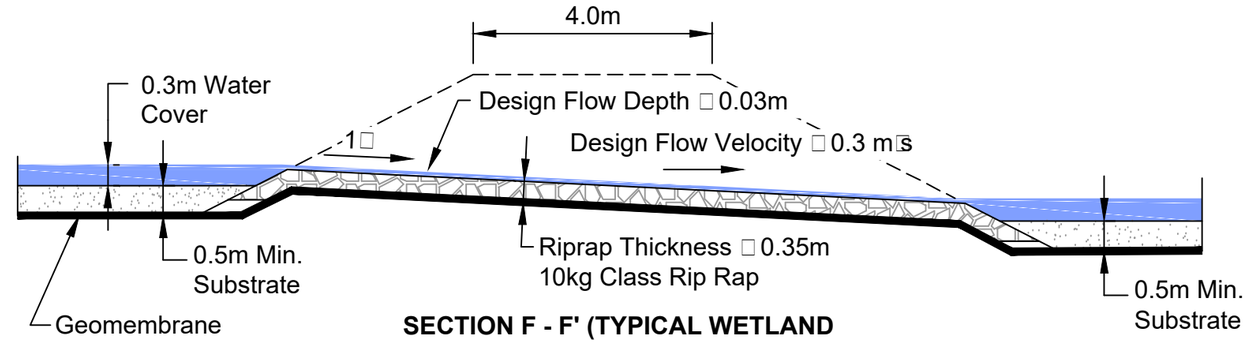
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TYPICAL CELL PLAN VIEW

0 5 10 15 20 25

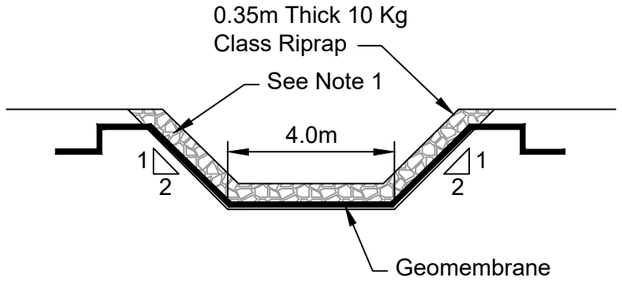
Scale in Metres



SECTION F - F' (TYPICAL WETLAND CELL CONVEYANCE CHANNEL)

F

NTS



SECTION G - G' (TYPICAL WETLAND CELL CONVEYANCE CHANNEL)

G

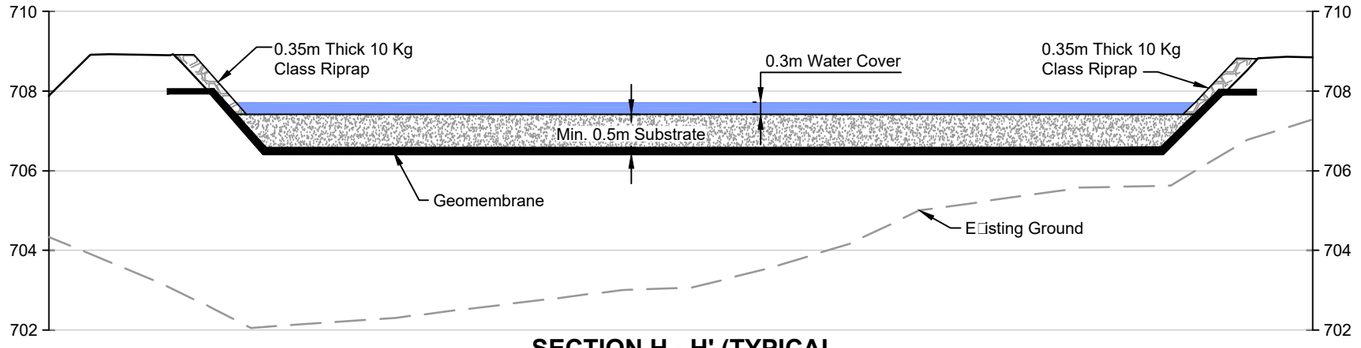
NTS

10kg CLASS - RIP RAP

Material Type	Rip Rap Dia. (mm)
D_{100}	300
D_{85}	280
D_{50}	200
D_{15}	90
Layer Thickness	350

100 kg CLASS - RIP RAP

Material Type	Rip Rap Dia. (mm)
D_{100}	680
D_{85}	610
D_{50}	450
D_{15}	200
Layer Thickness	700



SECTION H - H' (TYPICAL CONSTRUCTED WETLAND)

H

0 2 4 6 8 10

Horizontal Scale in Metres
Vertical Exaggeration 2

LEGEND

- Rip Rap
- Constructed Wetlands Design
- Geomembrane
- Flow Direction

NOTES

1. Cell Conveyance Channel requires a minimum 350 mm layer of rock rip-rap having an average diameter of greater than 200 mm to mitigate erosion.
2. Typical 4.0 m wide berms to be constructed to provide access for Construction and monitoring and maintenance.

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Preliminary Infrastructure Minto Creek Constructed Wetlands

DRAWING TITLE: Typical Cell and Cell Spillway Layout

DRAWING NO.	SHEET	REVISION NO.
CW-04	5 OF 5	A

Closure Water Conveyance System Design Update Report, Minto Mine

Prepared for

Minto Explorations Ltd.



Prepared by

SRK Consulting (Canada) Inc.
1CM002.042
February 2018

Closure Water Conveyance System Design Update Report, Minto Mine

February 2018

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Appendix A – Conveyance System Drawings

Appendix B – Site-Specific Hydrology Review

Appendix C – Climate Change Analysis

Appendix D – Channel Routing Options Evaluation

Appendix E – Integrating Vegetation Erosion Protection

1 Introduction

SRK Consulting (Canada) Inc. has prepared this updated post-closure water conveyance system design report for the Minto Mine located in Yukon, Canada, and operated by Minto Explorations Ltd. The updated designs are intended to form a part of Minto's January 2018 update of the mine's Reclamation and Closure Plan (RCP 2018-01).

The designs of the post-closure water conveyance system described herein are updates to those developed for the previous version of the RCP (RCP 2016-01). The updated designs account for the following:

- Updated as-built details of the current ground surface;
- Updated as-built and design details for the Main Pit Dump;
- Redesign of the ultimate Area 2 Stage 3 Pit; and
- Refinements based on outcomes of the 2017 Failure Modes and Effects Assessment (AEG 2018).

Because of the above revisions, the existing post-closure designs for the water conveyance network and related infrastructure required updates. Included in this report is the preliminary design level of the post-closure water network. Secondary and tertiary drainage features are not included in the preliminary engineering design for the conveyance network; however, the accompanying cover designs address local drainage networks as part of managing surface runoff and limiting erosion for covered facilities (SRK 2018).

The post-closure conveyance system is a network of channels, erosion protection features, and energy dissipation structures that convey surface water from and through the mine site to Minto Creek, which itself discharges into the Yukon River. This design report focuses on the primary arterial network of conveyance channels within the mine footprint. The network of channels is designed to collect, intercept, and convey water through the mine in its closed state. To support the conveyance system designs and methodology summarized in this report, the following technical memos are provided in the appendices:

- Appendix A – Conveyance System Drawings,
- Appendix B – Site-Specific Hydrology Review,
- Appendix C – Climate Change Analysis,
- Appendix D – Routing Flow Closure Evaluation, and
- Appendix E – Integrating Vegetation Erosion Protection.

2 Water Management Review

To support the water conveyance designs, SRK has reviewed and updated the site-wide hydrology analysis to reflect climate change, modified watershed catchment areas and assessed

options of alignments for conveyance channels. The results of this review are summarized below. Further details are provided in the technical memos (Appendices B, C and D).

2.1 Hydrology Review

This section summarizes the background and results of the site wide hydrology review which is included with this report in Appendix B.

The hydrology analysis was updated to estimate the summer probable maximum precipitation (PMP), the probable maximum flood (PMF), and 1:200-year peak flows for the design of the conveyance channels and erosion protection required for the Reclamation and Closure Plan (RCP).

The maximum instantaneous 200-year peak flow and PMF were calculated using a rainfall-runoff methodology where maximum average daily snowmelt was modelled as baseflow. The Soil Conservation Service method was used to determine runoff based on catchment area characteristics such as: lag time, area, and curve number coefficients. The Soil Conservation Service Type I rainfall distribution was selected for use in the HEC-HMS (2013) model. The Type I distribution approximates an intense, short-duration storm event typical of storms in the Yukon and nearby Alaskan interior. A HEC-HMS model was developed for the post-closure conveyance network, to determine the 1:200 year return-period peak flows and the PMF along the post-closure conveyance channels.

The PMP for a 24-hour duration event was evaluated to be 159 mm of precipitation based on observed data from 20 regional meteorological stations. Additional details of the hydrology and evaluation of the PMP are presented in Appendix B.

2.2 Climate Change Analysis

This section summarizes the background and results of the climate change study which is included in Appendix D of this report.

The International Panel on Climate Change (IPCC) projects a 5°C to 6°C increase in the ambient temperature in Northern Canada for year 2100 relative to the 1980 to 1999 temperatures (IPCC 2007). Evaluating the climate change models relative to Minto Mine, the mean annual air temperature has the potential to increase 3.3°C over the next century. Trending forward in time, the mean annual air temperatures may reach −0.8°C, 0.2°C, and 1.1°C in the 2020s, 2050s, and 2080s, respectively.

According to the climate change model, the total precipitation at Minto is forecasted to increase 67.9 mm (15%) during the closure and post-closure periods (2011 to 2040) and 199.3 mm (44%) by 2100. Heavy rainfall, which is the fraction of annual precipitation greater than the 95th percentile, is predicted to increase by 96% over the next century at Minto. Similarly, the number of days with more than 10 mm of rainfall is forecasted to increase by 140%. This trending increase in annual precipitation was accounted for in the sizing and specifications of conveyance structures (Section 3).

The monthly average snow depth at Minto is expected to increase over the next century. Forecasts suggest that freshet will occur in April, and maximum daily snowmelt will increase over the next century.

2.3 Channel Routing Options Evaluation

This section summarizes the results of an alternative assessment for the channel alignment around the Main Pit and Area 2 Stage 3 Pit. Channel alignments and flood routing through the pits were evaluated for a range of design storms including the PMF.

Two flood routing alternatives for the Area 2 Stage 3 Pit were considered:

- Option 1 – Area 2 Stage 3 Pit upstream flow is routed into the pit through Channel B. From the pit, flows are discharged to Channel D.
- Option 2 – Area 2 Stage 3 Pit upstream flow is routed around the pit through Channel B. Channel B and Channel C both report to Channel E.

The results of the two routing options illustrate a minimal difference in the magnitude of the PMF in downstream channels for Options 1 and 2. Furthermore, the Option 2 of diverting flows around the Area 2 Stage 3 Pit does not result in any measurable difference in the timing of the peak flow into lower Minto Creek.

A total of three geometric channel configurations were evaluated for the pit outlet channels to assess hydraulic design requirements (i.e. channel depth and erosion protection):

- Case 1 – Minimum spillway width (2 m).
- Case 2 – Constructible spillway width, based on dozer sizing (7 m).
- Case 3 – Maximum spillway width, based on topography (22 m for Main Pit, 15 m for Area 2 Pit).

Based on the routing alternatives and channel geometry assessment, it is recommended that channel alignments flow into the Main Pit and Area 2 Stage 3 Pit.

Based on the constructability methods and costs of using a dozer, it is recommended that channel widths be sized to a minimum of a dozer blade (i.e. Case 2). Given that there is no benefit to route flows around the pits, it is recommended that channel alignments flow into the pits and should be sized to have a minimum bottom width of a dozer blade (i.e. 7 m). More detail of this analysis is included in Appendix D.

3 Design Basis

The design basis of the closure water conveyance system is to collect upstream surface runoff and convey flows through the mine site to downstream Minto Creek.

The conveyance system should be designed in a manner that constructed water management features are robust and durable post-closure. Key design criteria considered in the design include:

- Rip rap erosion protection and integration of vegetation,
- Robust geometry that reduces the risk of overtopping and erosion,
- Allowance for future climate change conditions and discharge requirements, and
- Utilization of pit lakes for mixing, energy dissipation, and sediment deposition.

3.1 Summary of the Channel Alignments

The conveyance channel system is composed of seven primary conveyance channels, the Stilling Basin, and the Primary Head Pond (Drawing CC-01; Appendix A). These channel alignments were selected based on four design considerations:

- Ability to convey surface runoff generated from upstream catchment areas,
- Minimization of channel slopes and excavation volumes (while maintaining alignments that are robust against excess flows and negative effects of long term thaw consolidation of foundation materials),
- Utilization of pit lakes established in the Main Pit and Area 2 Stage 3 Pit to dissipate energy and settle sediments, and
- Convey flows generated on site to the Minto Creek channel upstream of the W3 monitoring station downstream of all mine disturbances.

The following section provides a summary of the function of each conveyance structure at closure.

- **Conveyance Channel A:** Located southwest of the Main Pit, downgradient of the Main Waste Dump (MWD) and the Southwest Waste Dump (SWD). This channel will convey surface water from the W15 tributary across the South Wall Buttress into the Main Pit lake. The channel alignment will incorporate a light vehicle swale crossing at its inlet to maintain access to the upgradient catchment areas.
- **Conveyance Channel B:** Located east of the Area 2 Stage 3 Pit. This channel collects flow from the Airport Road Ditch and from the W35A tributary and conveys into the Area 2 Stage 3 Pit. The channel is cut into the pit wall and incorporates erosion protection. The upstream channel alignment incorporates a light vehicle swale crossing to maintain access to the underground portal to the west of the channel (South Portal).
- **Conveyance Channel C:** Channel C conveys discharge water from the Main Pit to a downstream confluence with Channel D, where the two combine to form the upstream segment of Channel E. Additionally, a constructed access road is required to maintain access to Channel C and to the Main Pit for post-closure monitoring and maintenance.

- **Conveyance Channel D:** Channel D conveys the discharge water from the Area 2 Stage 3 Pit to the confluence with Channel C. The confluence of flows from Channel C and D requires additional design attention to ensure flow transitions don't result in channel erosion. Similar to Channel C, the channel design also includes a parallel access road for post-closure monitoring and maintenance.

- **Conveyance Channel E:** Channel E has been subdivided into a system of three sequential channels: E1, E2, and E3. Channel E ultimately conveys water from the Main and Area 2 Stage 3 open pits into the Constructed Wetland Treatment System (CWTS). Flows in excess of the design of the CWTS flow rate are directed into a high-flow bypass channel (Channel G).

  Channel E1 commences at the confluence between channels C and D and conveys water in a constructed channel at a slope grade of 1% to the existing alignment of the site access road. Channel E2 and E3 is constructed on the alignment of the existing access road with grades ranging from 1.5% to 20%. Downstream of Channel E3, high velocity flows are transitioned into the Stilling Basin located upstream of the Primary Head Pond.

- **Conveyance Channel F:** Conveyance Channel F has been subdivided into two channels. Channel F1 represents the existing Tailings Diversion Ditch and Channel F2 represents the downstream segment of the diversion to convey flows upstream of the Dry Stack Tailings Storage Facility to Channel G (High Flow Bypass Channel).

- **Conveyance Channel G:** The High Flow Bypass Channel conveys flows from the Primary and Secondary Head Ponds through the High Flow Control Structures around the CWTS to lower Minto Creek. Flows that are diverted around the wetland are those that are in excess of the design flow rate for the CWTS or which enter the channel via Channel F (SRK 2016 and Contango 2016).

- **Stilling Basin:** Downslope of Channel E3, high velocity water is received by a Stilling Basin which is designed to dissipate energy and contain the development of a hydraulic jump within the protected stilling basin surface. Water passes from the Stilling Basin into the Primary Head Pond.

- **Primary Head Pond:** The Primary Head Pond moderates turbulent flows that exit the Stilling Basin before water is directed into the CWTS or the High Flow Bypass Channel by flow control structures on the south side of each head pond. The dual head pond system is designed to step down the flow rate discharging the Channel E3 to essentially separate the wetland design flow from the remainder of the water discharging the site. A design layout for the Primary and Secondary head ponds are illustrated in the Minto Creek Constructed Wetland: Physical Infrastructure Preliminary Design report (SRK 2016).

3.2 Design Criteria

3.2.1 Channel Design Criteria

All closure conveyance channels are designed to be lined with riprap for erosion protection for flows up to a 24-hour, 200-year return period. Evaluation of the suitability of using vegetation as a stand-alone protection will be considered at a later stage of design. Further details on the selection of vegetation species and biotechnical techniques are included in Appendix E. Given that over time, vegetation will naturally establish within the conveyance system, channels were sized to convey the 200-year peak flow with vegetation growth.

For channels with cross-slope alignments that have a risk of overtopping during a return period greater than the 200-year event and causing significant downstream effect (i.e. Channel B), channels were designed to convey the PMF. Table 3-1 provides a summary of the design criteria applied for the erosion and conveyance (i.e. channel sizing).

Table 3-1: Channel Design Criteria Summary

Channel	Design Criteria		Description
	Erosion Protection	Conveyance	
A	200-year	PMF	Risk of overtopping onto Main Pit Dump and causing erosion. Channel sized to contain PMF.
B		PMF	Risk of overtopping onto Dry Stack Tailings Storage Facility and causing erosion. Channel sized to contain PMF.
C		200-year + freeboard	Surrounding topography provides adequate relief to convey PMF.
D		200-year + freeboard	Surrounding topography provides adequate relief to convey PMF.
E1		200-year + freeboard	Surrounding area graded to form valley-bottom to convey PMF to Channel E2.
E2		200-year + freeboard	Surrounding area graded to form valley-bottom to convey PMF to Channel E3.
E3		200-year + freeboard	Surrounding area graded to form valley-bottom to convey PMF to Stilling Basin.
F1		PMF	Risk of overtopping onto Dry Stack Tailings Storage Facility and causing erosion. Channel sized to contain PMF.
F2		200-year + freeboard	Surrounding topography provides adequate relief to convey PMF (PMF contained within natural gulley).
G		200-year + freeboard	PMF contained in the valley-bottom to downstream Minto Creek. Overtopping of wetland infrastructure will occur under flows that exceed channel capacity.

All closure conveyance channels are designed with a typical geometry that reflects the size and capability of the heavy equipment used on site. All channels were designed to have a minimum base width of 7 m (D11 dozer blade width), side slopes of 2:1 and a minimum freeboard of 0.3 m.

The base of the channel will be graded to create a thalweg in the center of the channel instead of incorporating a pilot channel as a design element. During low flow events, flows will be conveyed along the thalweg.

Excavated depths are variable due to channel grade and riprap sizes and each channel is divided into segments that relate to a typical riprap classification. A summary of required channel depths and rip rap specifications are provided in the drawing package (Appendix A).

The closure conveyance channels are designed to be lined with riprap for erosion protection and geotextile to prevent sediment migration and erosion of the channel side slopes. Sediment migration and erosion of the base material beneath the riprap protection may not warrant any geotextile along the proposed alignments, however geotextile has been incorporated at this stage of design. A geotextile element may not ultimately be necessary because the excavated material from which the channel alignments are constructed consists of either heavily compacted fills from existing haul roads or coarse rock fills generated during mine development. The need for geotextile will be considered further during later design.

Seepage losses from Channel E2 would ultimately report to the Stilling Basin, and are therefore expected to not represent a risk to closure performance. Seepage volumes are expected to be low, in part due to the highly compacted nature of the mine access road that forms the base of the post-closure channel. As such, no additional low permeability element has been incorporated into the design.

3.2.2 Stilling Basin Design Criteria

This structure is designed and protected to convey up to the PMF. Channel E3 enters the Stilling Basin at the western (upstream) end; Channel E2 has a maximum slope of 20% followed by a slope of 0.5% in the Stilling Basin itself. This abrupt slope change is anticipated to produce a hydraulic jump under design flows, and that hydraulic jump will provide energy dissipation. To improve the energy reduction, the channel width is increased from 7 to 20 m.

3.2.3 Primary Head Pond

The design criteria developed for the management of the passive treatment wetlands is to portion 3000 m^3/day of water from the discharge of E3 and the Stilling Basin and convey it into the wetlands (Contango 2016). To moderate the flows entering the CWTS and to direct excess flows to the High Flow Bypass Channel, a sequence of inlet flow control structures and outlet overflow structures are required.

The second component to the CWTS flow conveyance system is required to bypass the flows that exceed the design flow capacity of the CWTS. The High Flow Bypass Channel is designed to convey excess flow from the two head ponds around the wetland system, as well as to keep the discharge from the Tailings Diversion Ditch separate from the wetlands and route that water to lower Minto Creek. Downstream, wetland passive treatment discharge flows will join with the bypass flow and enter Minto Creek.

4 Methodology

4.1 Design Flow Estimates

Conveyance channel design flows were determined based on the site-wide hydrology and climate change reviews that were prepared by SRK (see Appendix B and C).

A rainfall-runoff approach was used to model peak flows. Rainfall was applied to a 24-hour distribution curve and was then converted to runoff based on catchment characteristics, including curve number, lag time, and total area, using the Soil Conservation Service (SCS) Method. A HEC-HMS model was developed to evaluate the peak 200-year and PMF to design the riprap protection and channel size of each conveyance channel.

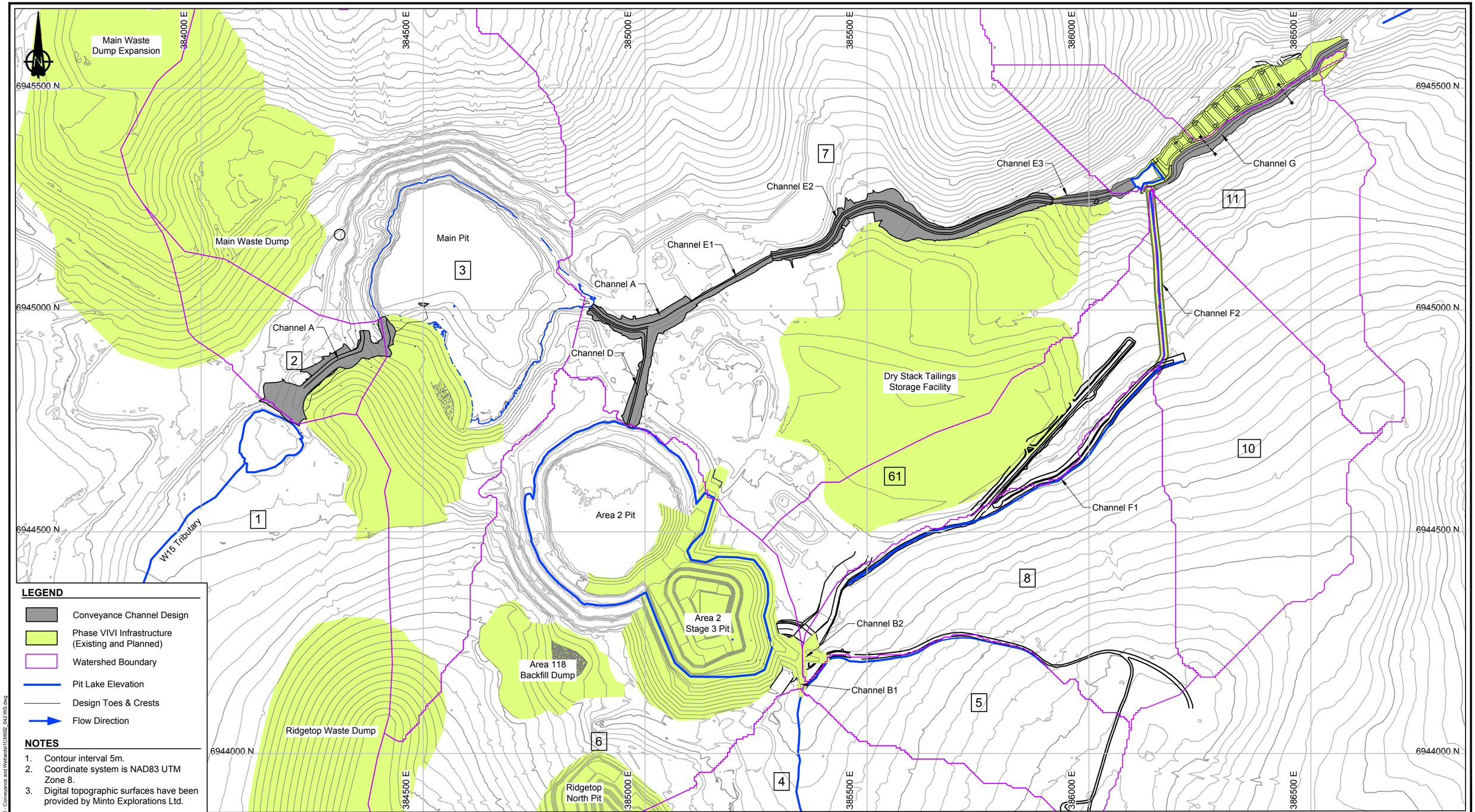
Figure 4-1 illustrates the catchment delineation for each channel and Table 4-1 summarizes the model inputs and Table 4-2 presents the 200-year and PMF flows for each channel. Appendix B provides additional details of the model inputs and validation.

Table 4-1: Summary of Hydrology Model Inputs

Catchment	Area (km²)	Lag Time (min)	Baseflow (m³/s)
1	3.46	27	0.92
2	0.10	6	0.03
3	0.86	6	0.23
4	1.87	21	0.50
5	0.17	9	0.05
6	0.64	7	0.17
7	1.81	17	0.33
8	0.32	18	0.09
9	0.21	19	0.06
10	0.32	12	0.09
11	0.29	12	0.08

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.042_ClosureConveyance\FMEA_2017\Minto - Channel_SPB.xlsm

Note: CN number applied to upstream watershed = 83



LEGEND

- Conveyance Channel Design
- Phase VIVI Infrastructure (Existing and Planned)
- Watershed Boundary
- Pit Lake Elevation
- Design Toes & Crests
- Flow Direction

NOTES

1. Contour interval 5m.
2. Coordinate system is NAD83 UTM Zone 8.
3. Digital topographic surfaces have been provided by Minto Explorations Ltd.

NOT FOR CONSTRUCTION

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srk consulting

capstone MINTO MINE OPERATED BY MINTO EXPLORATIONS LTD.

MINTO EXPLORATIONS Ltd.

DESIGN:	DRAWN: TH	REVIEWED:
CHECKED:	APPROVED:	DATE: 2018/01/12
FILE NAME: 1CM002_042-WS.dwg		SRK JOB NO.: 1CM002.042

Closure Conveyance Design

DRAWING TITLE:

Watershed Delineation

DRAWING NO.

4.1

Table 4-2: Summary of Peak Flows

Channel	200-year (m^3/s)	Probable Maximum Flood (m^3/s)
A	11.5	42.6
B	7.6	28.2
C	13.1	47.7
D	9.7	36.6
E1	28.9	110.1
E2	28.9	110.1
E3	28.9	110.1
F1	1.3	4.8
F2	2.3	8.6
G	31.5	121.3

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.042_ClosureConveyance\FMEA_2017\CatchmentSummary_Minto_Closure_2018_spb_rev01.xlsm

4.2 Hydraulic Analysis (Channel Dimension and Erosion Protection)

The hydraulic analysis and sizing of channels were determined using Manning's Equation under a steady state normal depth condition. As described in Section 3.2.1, rip rap was sized for the 200-year peak flow and channels were sized for either the 200-year peak flow or PMF depending on the downstream risk (see Table 3-1) and the assumption that post-closure vegetation will be established. The Manning's roughness coefficient for a channel with vegetation in literature is determined to be 0.1 (Bedient et al, 2008). Studies of willows in channel have also been completed where it was found that the Manning's roughness coefficient for Sandbar Willows ranged from 0.04 to 0.08 depending of the channel depth (Chen et al. 2009). For the sizing of the conveyance channels at Minto a roughness coefficient of 0.1 was applied for channels with vegetation. The procedure to evaluate the Manning's coefficient for the rip rap is an iterative process between water depth, Manning's roughness and rip rap size.

Riprap sizing and Manning's roughness values were derived through model iterations. The Manning's equation was evaluated using initial roughness values, and the results were extracted at each channel segment of different longitudinal slope. For each interval, the D_{50} of the riprap was calculated based on the velocity in the channel, and the corresponding Manning's roughness value was calculated based on the riprap sizing. This process was repeated until similar *n* values were obtained between model iterations.

4.2.1 Riprap Calculations

Closure Conveyance Channels

Riprap sizing for the closure conveyance channels was estimated using three expressions: Isbash (USACE 1991), Robinson et al. (1998), and Khan and Ahmad (2011). To be conservative, the maximum of the three estimates was used in the conveyance channel design.

Equation 2 presents the Isbash expression.

$$D_{50} = \frac{V^2}{C^2 2g(\frac{\gamma_s - \gamma_w}{\gamma_w})}$$ Eq. [2]

Where *V* is the average velocity (ft/s), *g* is gravitational acceleration (ft/s²), γ_s is the unit weight of riprap (lb/ft³), γ_w is the unit weight of water (lb/ft³), D_{50} is the median riprap size (ft), and C is the Isbach coefficient. The Isbach coefficient was set to 1.2 for low turbulence or bank protection.

Equations 3 and 4 present the Robinson et al (1998) expressions, which are based on the channel slope.

$$D_{50} = (\frac{Q}{9.76x10^{-7} * S^{-1.50}})^{\frac{1}{1.89}} \qquad S < 0.10$$ Eq. [3]

$$D_{50} = (\frac{Q}{8.07x10^{-6} * S^{-0.58}})^{\frac{1}{1.89}} \qquad 0.10 \leq S \leq 0.40$$ Eq. [4]

Where *Q* is the unit peak flow rate (m³/s/m), D_{50} is the median riprap size (m), and *S* is the slope (m/m).

Equation 5 presents the Khan and Ahmad (2011) expression.

$$D_{50} = 0.66\, x\, t^{0.58}\, x\, S^{0.22}\, x\, {C_u}^{-0.45}\, x\, {Q_f}^{0.22}$$ Eq. [5]

Where *t* is the riprap thickness (m), *S* is the channel slope (m/m), C_u is the coefficient of uniformity, and Q_f is the critical unit discharge (m³/s/m).

Stilling Basin and Head Pond

The erosion protection of the Stilling Basin and Head Pond were designed to be protected to convey the PMF resulting in large peak flows. The above three rip rap sizing expressions were complemented with Wittler and Abt, Mishra, and LaGasse et al. (Abt et al. 2013), which are expressions that are more suitable for evaluating larger sized riprap.

Because of the variability of the rip rap results, the riprap selected for the Stilling Basin and Head Pond was based on the average of these all six expressions (i.e. Isbash, Robinson et al., Khan and Ahmad, Wittler and Abt, Mishra, and LaGasse et al.), where the minimum and maximum values were removed from the average.

4.2.2 Manning's Roughness Estimation

Given a riprap size estimate, a Manning's roughness value (n) was then calculated based on three relationships by Strickler, Anderson, and Abt et al. (1988). The average *n* value obtained from the three expressions was selected. A smaller *n* value corresponds to a higher velocity and lower flow depth and vice versa; the average provides a conservative velocity and depth for the design.

Equation 6 presents the Strickler expression.

$$n = 0.0385{K_s}^{1/6} \quad \text{Eq. [6]}$$

Where *n* is the Manning's roughness value, and K_s is the particle diameter of the largest 15% of the riprap grain size distribution (ft).

Equation 7 presents the Abt et al. (1988) expression.

$$n = 0.0456(D_{50}\ x\ S)^{0.159} \quad \text{Eq. [7]}$$

Where *n* is the Manning's roughness value, S is the channel slope (ft/ft), and $D_{50\ is}$ the average riprap diameter (inches).

Equation 8 presents the Anderson expression.

$$n = 0.0395{D_{50}}^{1/6} \quad \text{Eq. [8]}$$

Where *n* is the Manning's roughness value, and D_{50} is the average riprap diameter (ft).

4.2.3 Riprap Classification and Gradation for Conveyance Channels

The riprap classifications (Table 4-3) were developed based on MOT BC (2006) to follow the design rationale described in this section for the closure channel segments. As stated before, the conveyance channels are composed of channel segments that have a typical profile, slope, and velocity profile. Each segment was assigned a riprap class and a layer thickness. The layer thickness is an aggregate of rock diameter types. Details pertaining to the riprap specifications for each segment are provided in the drawing package (Appendix A).

Table 4-3: Riprap Class Gradation

Class of Riprap [kg]	Riprap Thickness [mm]	Rock Gradation Percentage Larger than Given Rock Mass [kg]			Rock Gradation Percentage Smaller than Given Rock Diameter [mm]				
		85%	50%	15%	D_{15}	D_{30}	D_{50}	D_{85}	D_{100}
10	400	1	10	30	90	140	200	280	300
25	600	2.5	25	75	130	200	300	380	450
100	900	10	100	300	200	300	450	610	680
250	1200	25	250	750	270	400	600	820	900
500	1600	50	500	1500	340	540	800	1030	1200
1000	2000	100	1000	3000	420	670	1000	1300	1500

5 Closure Conveyance Channel Design

This section summarizes the closure conveyance channel design parameters. Channel depths are variable and related to the depth of water being conveyed. Specification tables for the channel geometry, flow depths and riprap are provided in Appendix A. All conveyance channel alignments are selected to avoid sharp bends to minimize the risk of scour and erosion and ultimately long-term maintenance requirements. Additional design elements that are specific to each channel are provided in the following sections and in the drawing package included in Appendix A. Drawing CC-01 (Appendix A) illustrates the general post-closure arrangement of the primary water conveyance network.

The following designs are prepared as an interim surface water conveyance design for closure which is based on the as-built configuration of the August 2017 surface. The closure design will need to be reviewed and updated based on future changes to mining and closure concepts.

5.1 Channel A

The closure design concept of Channel A is to collect and convey surface water from the W15 tributary to the Main Pit lake. Surface water from the W15 tributary is designed to enter Channel A through an intake transition section. This section will be designed to a detail level upon the completion mining, using the final surface as-built details.

The alignment, profile and typical cross-sections for Channel A are illustrated in Drawing CC-02. Table 5-1 summarizes the hydraulic design for Channel A where the maximum slope is 35%. The channel was sized to convey the PMF (42 m^3/s) and rip rap protection was sized for the 200-year peak flow (11 m^3/s).

Table 5-1: Design Summary for Closure Conveyance Channel A

Segment ID	Length [m]	Slope [%]	Channel Base Width (m)	Minimum Channel Depth [m]	1:200 year Water Depth[1] [m]	Riprap	
						D50 (mm)	Class (kg)
1	83	0.091	10	1.50	0.55	300	25
2	159	0.040	10	1.50	0.70	200	10
3	107	0.353	10	1.00	0.37	450	100

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.042_ClosureConveyance\FMEA_2017\Minto - Channel_SPB.xlsm

Notes:

[1] Water depths represent the water depth for the 200-year peak flow with vegetation established post-closure in the channel.

5.2 Channel B

Currently routine flows from the W35A Tributary and Airport Diversion Ditch flow through a pipe crossing to the Tailings Diversion Ditch (Channel F). Flows in excess are routed through a spillway into the Area 2 Stage 3 Pit. The proposed closure concept for Channel B is to reroute all flows from W35A tributary and Airport Diversion Ditch to the Area 2 Stage 3 pit through a new alignment cut into the pit wall and sloped back at a 16% grade. The channel will be protected against erosion with rip rap and provide a thermal layer to reduce permafrost degradation.

The closure conveyance channel design will plug and backfill the entrance to the existing pipe crossing and regrade the current spillway to convey flows to the Channel B at closure. Surface water from the W35A tributary and the Airport Diversion Ditch is designed to enter Channel B through an intake transition section. These sections will be designed to a detail level upon the completion of final closure mine plan and will include rock riprap erosion protection and compacted earth berms that direct flows from the W35A tributary and the Airport Diversion Ditch into Channel B.

Drawings CC-03 illustrate the alignment, profile and typical cross-sections. The Table 5-2 summarizes the closure conveyance hydraulic design Channel B. The channel was sized to convey the PMF (28 m^3/s) and rip rap protection was sized for the 200-year peak flow (8 m^3/s).

Table 5-2: Design Summary for Closure Conveyance Channel B

Segment ID	Length [m]	Slope [%]	Channel Base Width (m)	Minimum Channel Depth [m]	1:200 year Water Depth[1] [m]	Riprap	
						D50 (mm)	Class (kg)
1	72	0.046	7	1.35	0.64	200	10
2	52	0.064	7	1.23	0.59	300	25
3	83	0.162	7	1.00	0.45	450	100
4	56	0.162	7	1.00	0.45	450	100

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.042_ClosureConveyance\FMEA_2017\Minto - Channel_SPB.xlsm

Notes:

[1] Water depths represent the water depth for the 200-year peak flow with vegetation established post-closure in the channel.

5.3 Channel C

The closure design concept for Channel C is to convey discharge water from the Main Pit water cover at elevation 786 m to the downstream confluence of Channel C (Area 2 Stage 3 Pit outlet channel). The intake structure in the Main Pit of Channel C will be designed to a detail level upon the completion of final closure plan.

Parallel to Channel C, a 4-m wide light vehicle access road will be designed. This road will provide post-closure access for monitoring and maintenance of the conveyance channel. The traffic surface of the access road is sloped towards the conveyance channel for drainage and to minimize excavation.

Drawings CC-04 illustrate the alignment, profile and typical cross-section for Channel C. Table 5-3 summarizes the closure conveyance hydraulic design for Channel C where a minimum channel slope of 1% was applied. The riprap erosion protection and channel was sized to convey the 200-year peak flow (13 m^3/s). Given the large cut, and parallel access road, it is also able to convey the PMF (48 m^3/s).

Table 5-3: Design Summary for Closure Conveyance Channel C

Segment ID	Length [m]	Slope [%]	Channel Base Width (m)	Minimum Channel Depth [m]	1:200 year Water Depth[1] [m]	Riprap	
						D50 (mm)	Class (kg)
1	182	0.010	7	1.70	1.35	200	10

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.042_ClosureConveyance\FMEA_2017\Minto - Channel_SPB.xlsm

Notes:

[1] Water depths represent the water depth for the 200-year peak flow with vegetation established post-closure in the channel.

5.4 Channel D

The closure design concept for Channel D is to convey discharge water from the Area 2 Stage 3 Pit at an elevation of 799 m to the downstream confluence of Channel C. The intake structure in from the Area 2 Stage 3 Pit to Channel D will be designed to a detail level upon the completion of final closure plan. A flow contraction transition structure constructed from overburden backfill and coarse oversize rock riprap is recommended for consideration to reduce permafrost degradation and direct the pit discharge into Channel D.

Drawings CC-05 illustrate the alignment, profile and typical cross-section for Channel D. Table 5-4 summarizes the closure conveyance hydraulic design. The riprap erosion protection and channel was sized to convey the 200-year peak flow (10 m^3/s). Given the large cut, and parallel access road, it is also able to convey the PMF (37 m^3/s).

Table 5-4: Design Summary for Closure Conveyance Channel D

Segment ID	Length [m]	Slope [%]	Channel Base Width (m)	Minimum Channel Depth [m]	1:200 year Water Depth[1] [m]	Riprap	
						D50 (mm)	Class (kg)
1	160	0.023	7	2.10	1.80	200	10
2	78	0.137	7	1.40	1.10	450	100

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.042_ClosureConveyance\FMEA_2017\Minto - Channel_SPB.xlsm

Notes:

[1] Water depths represent the water depth for the 200-year peak flow with vegetation established post-closure in the channel.

5.5 Channel E

Channel C and D converge to form Channel E. This confluence location was selected based on existing topography and the chosen channel grades for Channel C and D. The confluence will be designed to a detail level upon the completion of the final closure plan and will require an apron sufficiently large enough to protect against an increase in water depth and high velocities due to the development of a hydraulic jump. Channel E is composed of three sequential subsections E1, E2, and E3 and conveys flows from the mine site (i.e. Main Pit, Area 2 Stage 3 Pit and Dry Stack Tailings Storage Facility to the CWTS).

Channel E1 starts at the confluence of channels C and D and conveys flows originating from the Main Pit and from the Area 2 Stage 3 Pit and transitions into Channel E2 and E3 along the existing access and discharges into a stilling basin prior to flowing into the CWTS.

The Channel E alignment is designed to overlay the current mine access road. A light vehicle access road is designed parallel to the alignment. The existing road surface was chosen for construction of a channel because the roadway has undergone traffic compaction over the life of mine and is expected to experience minimal long-term settlement and minimal seepage interception from the Dry Stack Tailings Storage Facility. At future stages of closure design, the need for seepage management along Channel E2 should be evaluated and, if found necessary, may require additional design details to address the need to further limit seepage interception into the channel. This potential additional design requirement could be addressed through lining the channel or components of the channel with low conductivity geomembranes. At this stage of the design, the compacted road material of the existing site access road is accepted to have sufficient low permeability.

Drawings CC-06, CC-07 and CC-08 illustrate the alignment, profile and typical cross-sections for Channel E1, E2 and E3 respectively. Channel E1 and E1 have segments with slopes ranging from 1% to 8%. Channel E3 is designed as a high velocity conveyance channel with a slope of 20% into the Stilling Basin.

Table 5-5 summarizes the hydraulic design for Channel E. The riprap erosion protection and channel was sized to convey the 200-year peak flow (29 m^3/s). The mine site will be graded to create a valley bottom to direct all flows from the site into Channel E and ultimately the Stilling Basin. As such, the Channel E valley-bottom can convey the PMF (110 m^3/s).

Table 5-5: Design Summary for Closure Conveyance Channel E

Segment ID	Length [m]	Slope [m/m]	Channel Base Width (m)	Minimum Channel Depth [m]	1:200 year Water Depth[1] [m]	Riprap	
						D50 (mm)	Class (kg)
1	278	0.011	7	2.00	1.82	200	10
2	260	0.062	7	2.00	1.26	450	100
3	98	0.014	7	2.00	1.89	450	100
4	92	0.058	7	2.00	1.28	450	100
5	92	0.063	7	2.00	1.26	450	100
6	105	0.041	7	2.00	1.41	450	100
7	33	0.076	7	2.00	1.19	600	250
8	46	0.051	7	2.00	1.33	450	100
9	150	0.207	7	2.00	0.90	1000	1000

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.042_ClosureConveyance\FMEA_2017\Minto - Channel_SPB.xlsm

Notes:

[1] Water depths represent the water depth for the 200-year peak flow with vegetation established post-closure in the channel.

5.6 Stilling Basin

Downslope of the E3 channel is the Stilling Basin which is designed to contain the development of a hydraulic jump within the protected stilling basin surface. The Stilling Basin geometry consists of a 20 m wide basin channel with 2H:1V side slopes constructed from riprap. Outflow from the Stilling Basin transitions into the Primary Head Pond which has a surface elevation of 715 m.

Table 5-6 summarizes the hydraulic design for the Stilling Basin.

Table 5-6: Design Summary for the Stilling Basin

Segment ID	Length [m]	Slope [m/m]	Channel Base Width (m)	Minimum Channel Depth [m]	1:200 year Water Depth[1] [m]	Riprap	
						D50 (mm)	Class (kg)
10	40	0.005	20	3.30	1.50	450	100

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.042_ClosureConveyance\FMEA_2017\Minto - Channel_SPB.xlsm

Notes:

[1] Water depths represent the water depth for the 200-year peak flow with vegetation established post-closure in the channel.

5.7 Channel F (Tailings Diversion Ditch)

Channel F is composed of two sequential subsections. Channel F1 illustrates the existing Tailings Diversion Ditch (TDD) constructed to divert flows around the Dry Stack Tailings Storage Facility and transitions to Channel F2 down an existing natural gulley down to the High Flow Bypass Channel (Channel G).

The grade along the existing TDD (Channel F1) range from 0.8% to 7% and the channel dimensions fluctuate along the alignment. Drawings CC-09 illustrates the alignment, profile and a cross-section of Channel F1. This cross-section was chosen to illustrate the most critical section (i.e. flattest slope and shallow channel depth) along the alignment. Channel F1 has sufficient capacity to convey the post-closure 200-year peak flow (1.3 m^3/s) with a Manning's coefficient of 0.1 assuming vegetation will be established within the channel section. The flattest segment of Channel F1 (0.8%) has the capacity to convey the PMF (4.8 m^3/s) at a maximum channel depth of 1.4 m.

Channel F2 is aligned in an existing natural gulley with grades ranging from 8% to 30% near the outlet into Channel G. Drawing CC-10 illustrates the alignment, profile and a typical cross-section of the Channel F2 design. Channel F2 has a v-shaped cross-section with a 12% side-slope and a minimum channel depth of 0.8 m to convey the 200-year peak flow under post-closure conditions (i.e. with vegetation established within the channel). The riprap protection was designed for the 200-year peak flow. Table 5-7 summarizes the hydraulic design for Channel F2. Given that the alignment follows natural topography, the PMF (8.6 m^3/s) will not create a new flow path. Some long-term settlement along the channel is expected due to thawing of the underlying permafrost; however, as the channel follows an existing natural channel with a steep gradient, any differential settlement is not likely to result in a change in flow path.

Table 5-7: Design Summary for Closure Conveyance Channel F2

Segment ID	Slope [m/m]	Minimum Channel Depth [m]	1:200 year Water Depth[1] [m]	Riprap	
				D50 (mm)	Class (kg)
1	0.078	0.8	0.50	800	500
2	0.121	0.8	0.46	800	500
3	0.159	0.8	0.44	800	500
4	0.092	0.8	0.49	800	500
5	0.163	0.8	0.44	800	500
6	0.168	0.8	0.43	800	500
7	0.174	0.8	0.43	800	500
8	0.284	0.8	0.39	800	500
9	0.270	0.8	0.40	800	500

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.042_ClosureConveyance\FMEA_2017\Minto - Channel_SPB.xlsm

Notes:

[1] Water depths represent the water depth for the 200-year peak flow with vegetation established post-closure in the channel.

5.8 Channel G (High Flow Bypass Channel)

The High Flow Bypass Channel conveys flows from Channel F and excess flows the Head Pond of the CWTS. Excess flows from the head pond flow through a 40 m wide and 0.5 m deep spillway. Flows that are diverted around the CWTS are those that are in excess of the design flow rate that can be managed by the wetlands (SRK 2016 and Contango 2016).

Drawing CC-11 illustrates the alignment, profile and typical cross-sections for Channel F. Table 5-8 summarizes the hydraulic design for Channel G. The channel slope ranges from 1% to 5%. The riprap protection and channel was sized to convey the 200-year peak flow (32 m^3/s). The 200-year water depth post closure with vegetation established in the channel will vary along the alignment depending on topography. During the PMF event, water levels will rise and flow into the CWTS and ultimately downstream to Minto Creek.

Table 5-8: Design Summary for Closure Conveyance Channel G

Segment ID	Length [m]	Slope [m/m]	Channel Base Width (m)	Minimum Channel Depth [m]	Riprap	
					D50 (mm)	Class (kg)
1	28	0.01	7	2.5	300	25
2	62	0.05	7	2.0	300	25
3	61	0.02	7	2.0	300	25
4	84	0.03	7	2.0	300	25
5	334	0.03	7	2.0	300	25

Source: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.042_ClosureConveyance\FMEA_2017\Minto - Channel_SPB.xlsm

Notes:

[1] Water depths represent the water depth for the 200-year peak flow with vegetation established post-closure in the channel.

5.9 Primary Head Pond

The Primary Head Pond moderates turbulent flows that exit the Stilling Basin before water is directed into a secondary head pond of the CWTS or Channel G by two flow control structures (CC-11).

The dual head pond system is designed to step down the flow rate discharging from the E Conveyance Channel to protect the constructed wetland treatment system from excess flows.

Design details for the CWTS and the head pond flow control structures can be found in *Minto Creek Constructed Wetland: Physical Infrastructure Preliminary Design* (SRK 2016).

6 Information Required to Advance the Channel Design

To further develop the closure conveyance channel design towards a final closure design plan, the following data requirements and project developments are required:

- The final post-mining as-built surfaces (such as as-built waste rock, pit, and general mine area surfaces) will be required to update and optimize the conveyance channel design.
- Review site wide closure progress and closure designs for other site components. This will allow determination of design gaps or redundancies in the closure conveyance channel design and will permit identification of opportunities for optimization and integration.
- In conjunction with review of site-wide closure designs, surficial drainage features from adjacent areas will need to be incorporated into the conveyance system. Designed drainage features include elements in the final cover designs, landform grading on waste rock dumps and tailings covers, and mine elements (e.g. haul roads, pit shells, and the underground portal).
- Design elements for the planned constructed wetland treatment system within the current WSP footprint will need to be integrated.
- Include an as-built design of the Tailings Diversion Ditch upon its completion to finalize the design of the High Flow Bypass Channel and related wetland infrastructure.
- Final post closure road layout design will be required to ensure integration of the road and channel networks.
- It is expected that the channels will have limited seepage losses and that there is no requirement for incorporation of low permeability elements in the channel designs. This should be confirmed during final design.
- At future stages of closure design, details of the specific hydraulic design of the Main and Area 2 Stage 3 open pit outlets and the energy dissipation aprons will need to be developed.

This report, "Closure Water Conveyance System Design Update Report, Minto Mine", was prepared by

Sarah Portelance, MEng, PEng
Senior Consultant

PROFESSIONAL ENGINEER YUKON TERRITORY VICTOR ALEJANDRO MUNOZ SAAVEDRA

2/6/2018

Victor Muñoz S., MEng, PEng
Senior Consultant

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and reviewed by

Dylan MacGregor, MASc, PGeo
Principal Consultant

All data used as source material plus the text, tables, figures, and attachments of this document have been reviewed and prepared in accordance with generally accepted professional engineering and environmental practices.

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The opinions expressed in this report have been based on the information available to SRK at the time of preparation. SRK has exercised all due care in reviewing information supplied by others for use on this project. Whilst SRK has compared key supplied data with expected values, the accuracy of the results and conclusions from the review are entirely reliant on the accuracy and completeness of the supplied data. SRK does not accept responsibility for any errors or omissions in the supplied information, except to the extent that SRK was hired to verify the data.

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Appendix A – Conveyance System Drawings

Engineering Drawings for the Closure Conveyance Design, Minto Mine, Yukon, Canada

ACTIVE DRAWING STATUS

DWG NUMBER	DRAWING TITLE	REVISION	DATE	STATUS	REPLACED REVISIONS
CC-00	Engineering Drawings for the Closure Conveyance Design, Minto Mine, Yukon, Canada	B	Jan. 16, 2018	Issued For Review	July 27, 2016; Rev A IFR
CC-01	Post-Closure General Arrangement	B	Jan. 16, 2018	Issued For Review	July 27, 2016; Rev A IFR
CC-02	Channel A	B	Jan. 16, 2018	Issued For Review	July 27, 2016; Rev A IFR
CC-03	Channel B	B	Jan. 16, 2018	Issued For Review	July 27, 2016; Rev A IFR
CC-04	Channel C	B	Jan. 16, 2018	Issued For Review	July 27, 2016; Rev A IFR
CC-05	Channel D	B	Jan. 16, 2018	Issued For Review	July 27, 2016; Rev A IFR
CC-06	Channel E1	B	Jan. 16, 2018	Issued For Review	July 27, 2016; Rev A IFR
CC-07	Channel E2	B	Jan. 16, 2018	Issued For Review	July 27, 2016; Rev A IFR
CC-08	Channel E3 AND STILLING BASIN	B	Jan. 16, 2018	Issued For Review	July 27, 2016; Rev A IFR
CC-09	Channel F1 Existing Tailings Diversion Ditch	F	Jan. 16, 2018	Issued For Review	
CC-10	Channel F2	F	Jan. 16, 2018	Issued For Review	
CC-11	Channel G High Flow By-Pass	B	Jan. 16, 2018	Issued For Review	July 27, 2016; Rev A IFR

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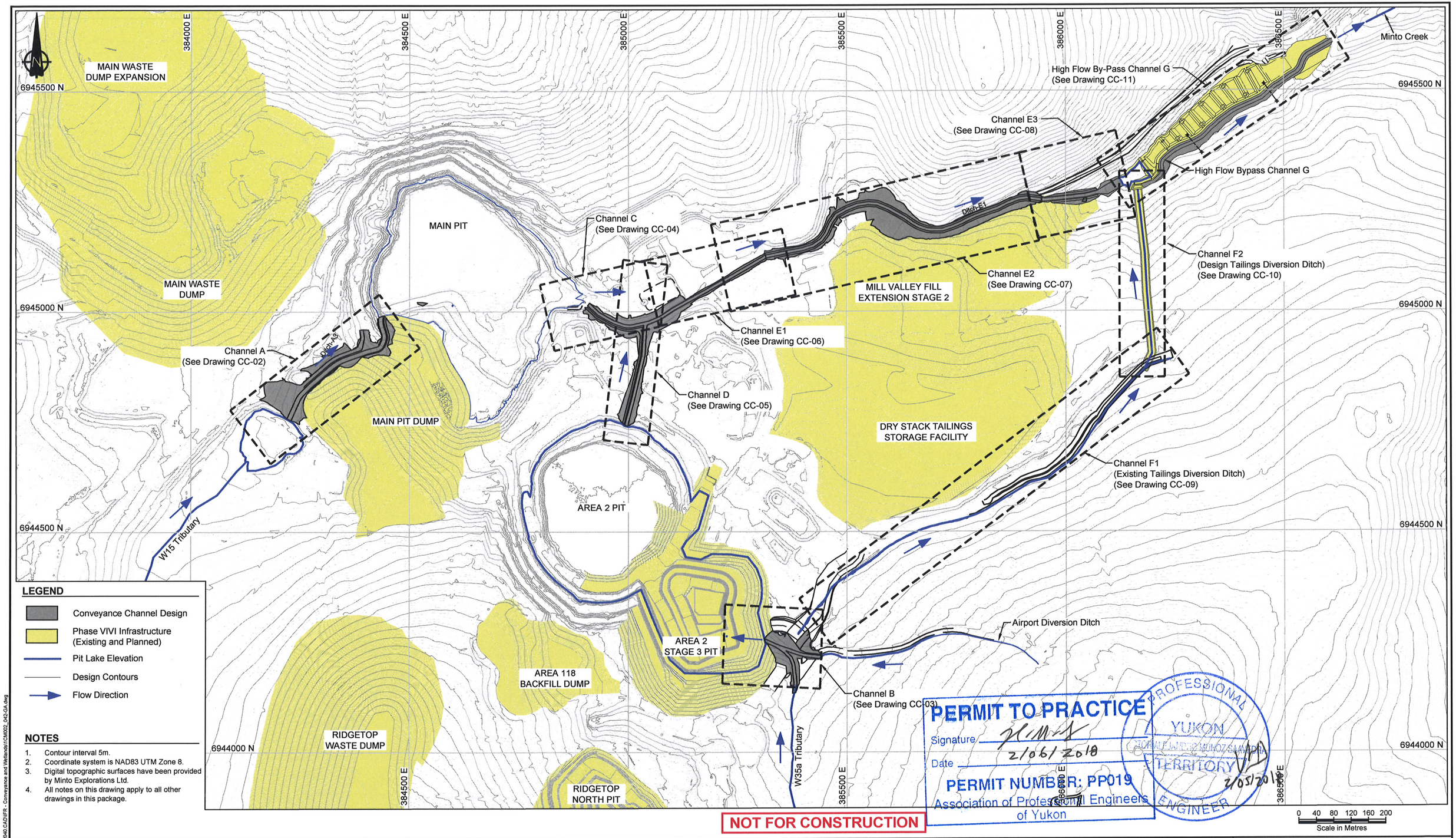
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Revision B - Issued for Review
January 16, 2018
Drawing CC-00



LEGEND

- Conveyance Channel Design
- Phase VIVI Infrastructure (Existing and Planned)
- Pit Lake Elevation
- Design Contours
- Flow Direction

NOTES

1. Contour interval 5m.
2. Coordinate system is NAD83 UTM Zone 8.
3. Digital topographic surfaces have been provided by Minto Explorations Ltd.
4. All notes on this drawing apply to all other drawings in this package.

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A	Issued for Review	SM	VM	15Jul16

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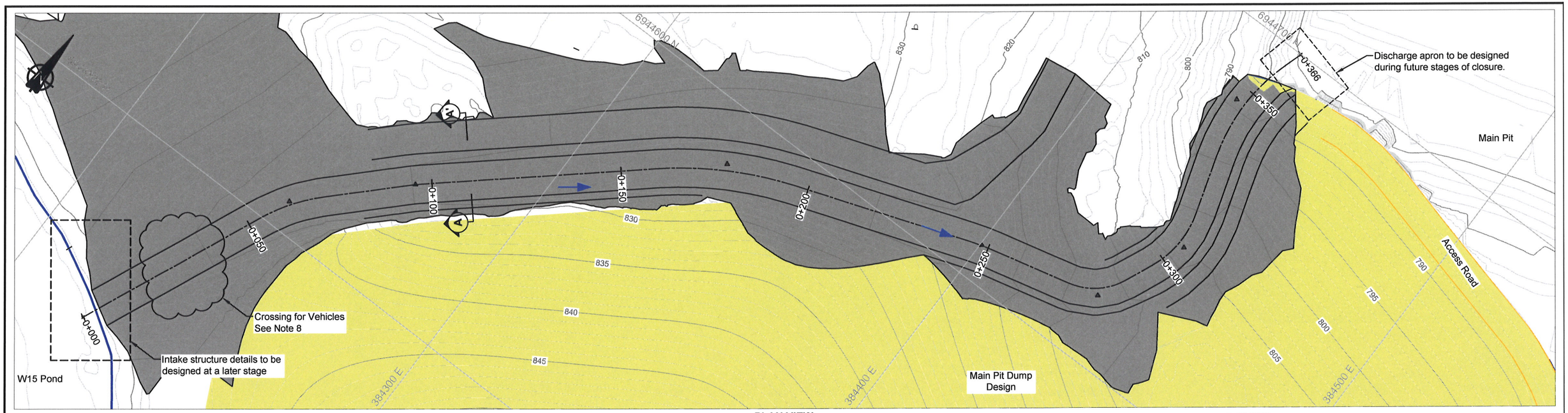
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Closure Conveyance Design

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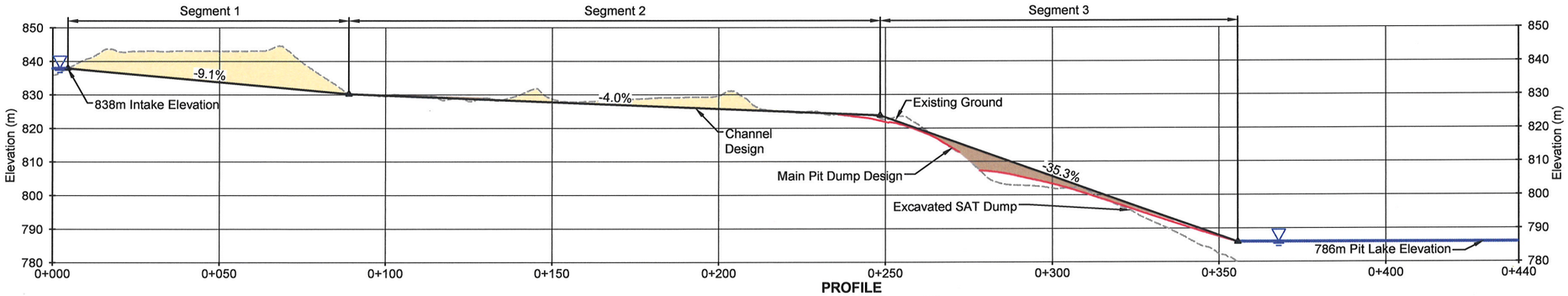
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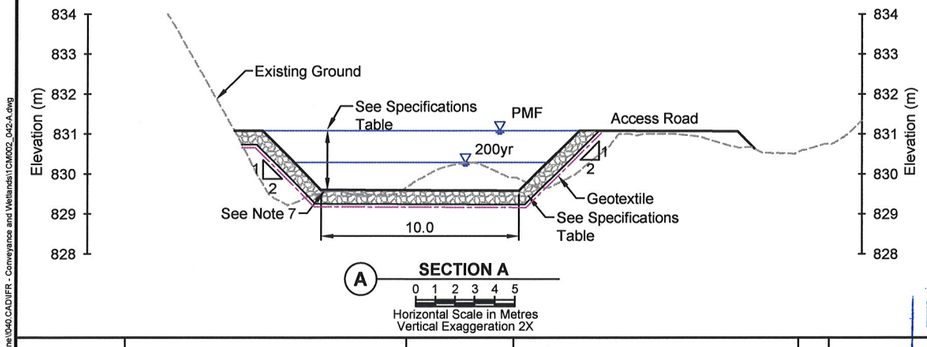
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PLAN VIEW
A CONVEYANCE CHANNEL

0 5 10 15 20 25
Scale in Metres





PROFILE
A CONVEYANCE CHANNEL

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Scale in Metres

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ALEJANDRO MUNOZ SAAVEDRA
TERRITORY
2/05/2018

LEGEND

- Conveyance Channel Design
- Channel Fill
- Channel Excavation
- Phase VIVI Infrastructure (Existing and Planned)
- Pit Lake Elevation
- Design Toes and Crests
- Flow Direction

NOTES

1. Contour interval 1m.
2. Digital topographic surfaces have been provided by Minto Explorations Ltd.
3. Topographic surfaces shown are comprised of existing and designed surfaces.
4. Constructed channel specifications are provided in the specifications table as per each channel segment.
5. Rip Rap classification gradation tables are provided in Drawing CC-06.
6. Future mining activities may alter the topography of the channel alignment. SRK recommends that at future stages of closure design channel alignment and grades be re-assessed.
7. The base of the channel will be graded to create a thalweg in the centerline.
8. A light vehicle crossing is to be designed into the closure conveyance channel upon the completion of the channel construction. A swale crossing with 5:1 side slopes should be considered during detailed design and alignment selection.

CHANNEL A SPECIFICATIONS:

Segment	Length (m)	Slope (%)	Minimum Channel Depth (m)	Channel Bottom Width (m)	Water Depth for 1:200 yrs (m)	Rip-rap Class (kg)	Rip-rap Thickness (mm)	D_{50} (mm)
1	83	-9.1	1.5	10.0	0.55	25	600	300
2	159	-4.0	1.5	10.0	0.70	10	400	200
3	107	-35.3	1.0	10.0	0.37	100	900	450

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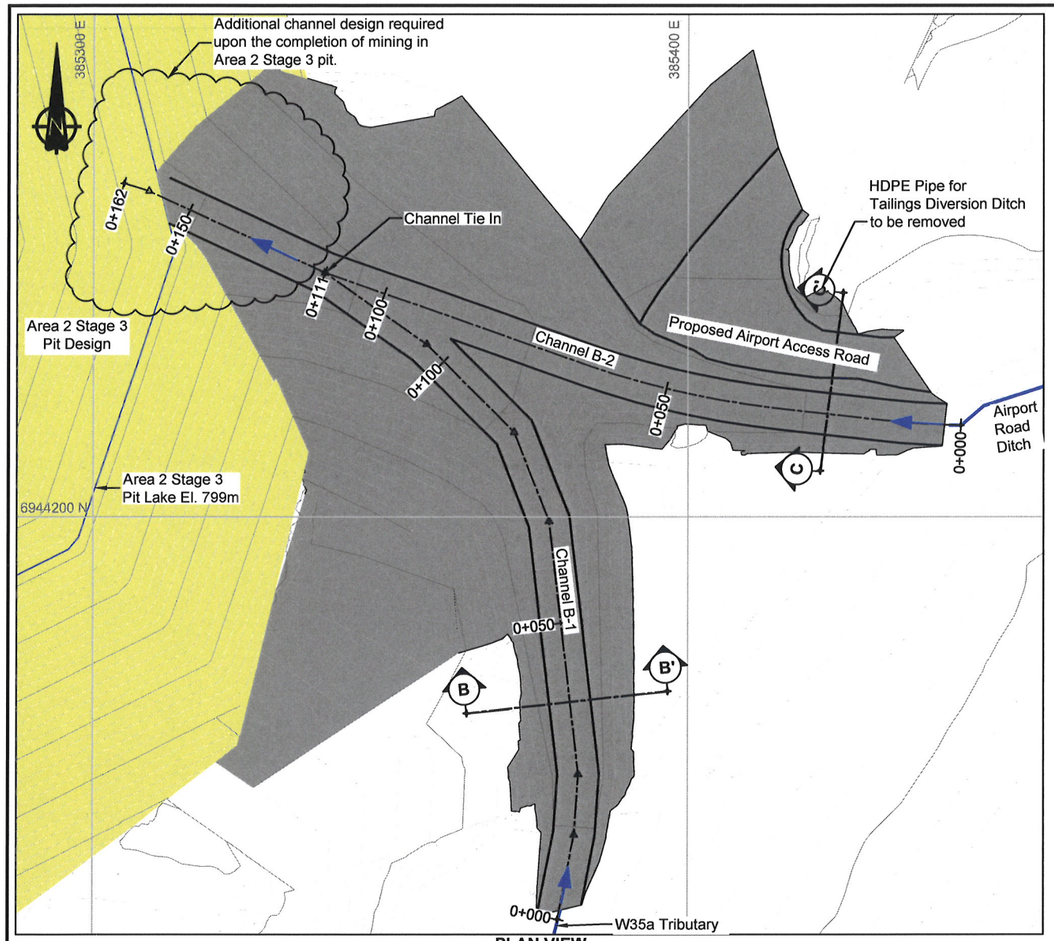
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Closure Conveyance Design

DRAWING TITLE: Channel A

DRAWING NO. CC-02 | SHEET 3 OF 12 | REVISION NO. B

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PLAN VIEW
B CONVEYANCE CHANNEL

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Scale in Metres

LEGEND

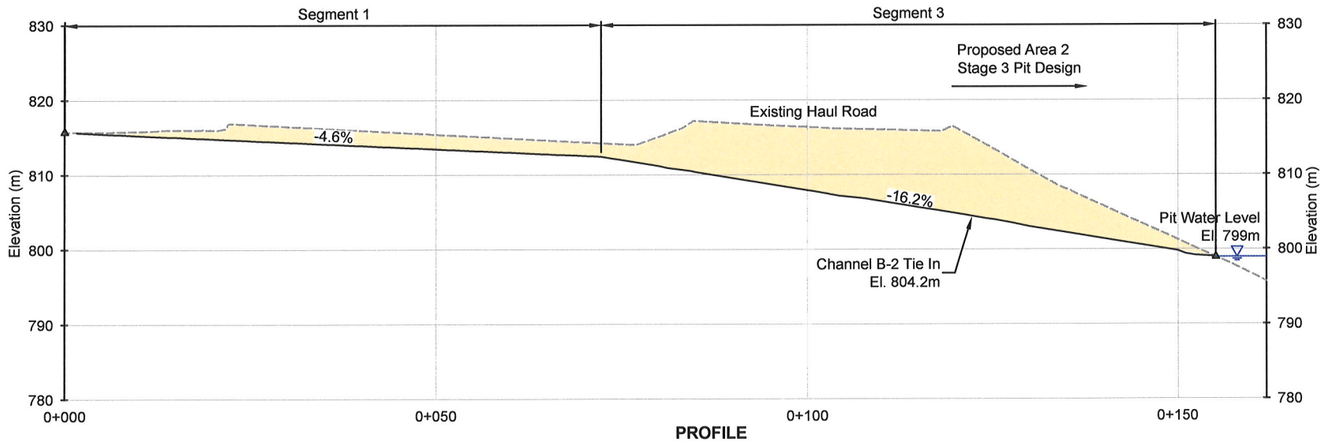
- Conveyance Channel Design
- Channel Fill
- Channel Excavation
- Phase VIVI Infrastructure (Existing and Planned)
- Pit Lake Elevation
- Access Roads, Design Toes & Crests
- Flow Direction

NOTES

1. Contour interval 1m.
2. Digital topographic surfaces have been provided by Minto Explorations Ltd.
3. Topographic surfaces are comprised of existing and designed surfaces.
4. Swale crossing design for light vehicle access to be designed later.
5. Rip Rap classification gradation tables are provided in Drawing CC-06.
6. The base of the channel will be graded to create a thalweg in the centerline.

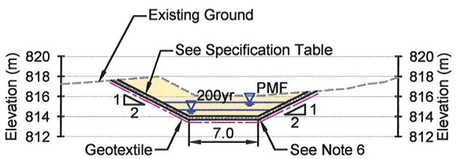
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Segment	Length (m)	Slope (%)	Minimum Channel Depth (m)	Channel Bottom Width (m)	Water Depth for 1:200 yrs (m)	Rip-rap Class (kg)	Rip-rap Thickness (mm)	D_{50} (mm)
1	72	-4.6	1.35	7.00	0.64	10	400	200
2	52	-6.4	1.23	7.00	0.59	25	600	300
3	83	-16.2	1.00	7.00	0.45	100	900	450
4	56	-16.2	1.00	7.00	0.45	100	900	450



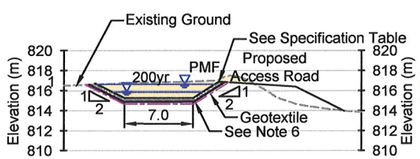
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B-1 CONVEYANCE CHANNEL

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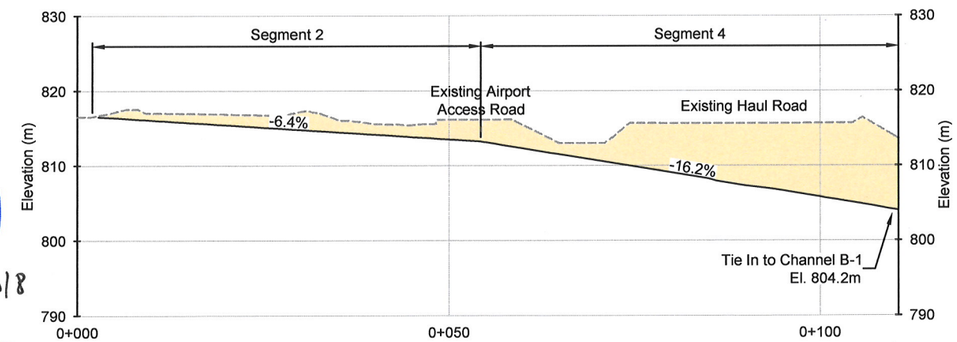
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TYPICAL SECTION

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Horizontal Scale in Metres



SECTION C
TYPICAL SECTION

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Horizontal Scale in Metres



PROFILE
B-2 CONVEYANCE CHANNEL

0 4 8 12 16 20
Scale in Metres

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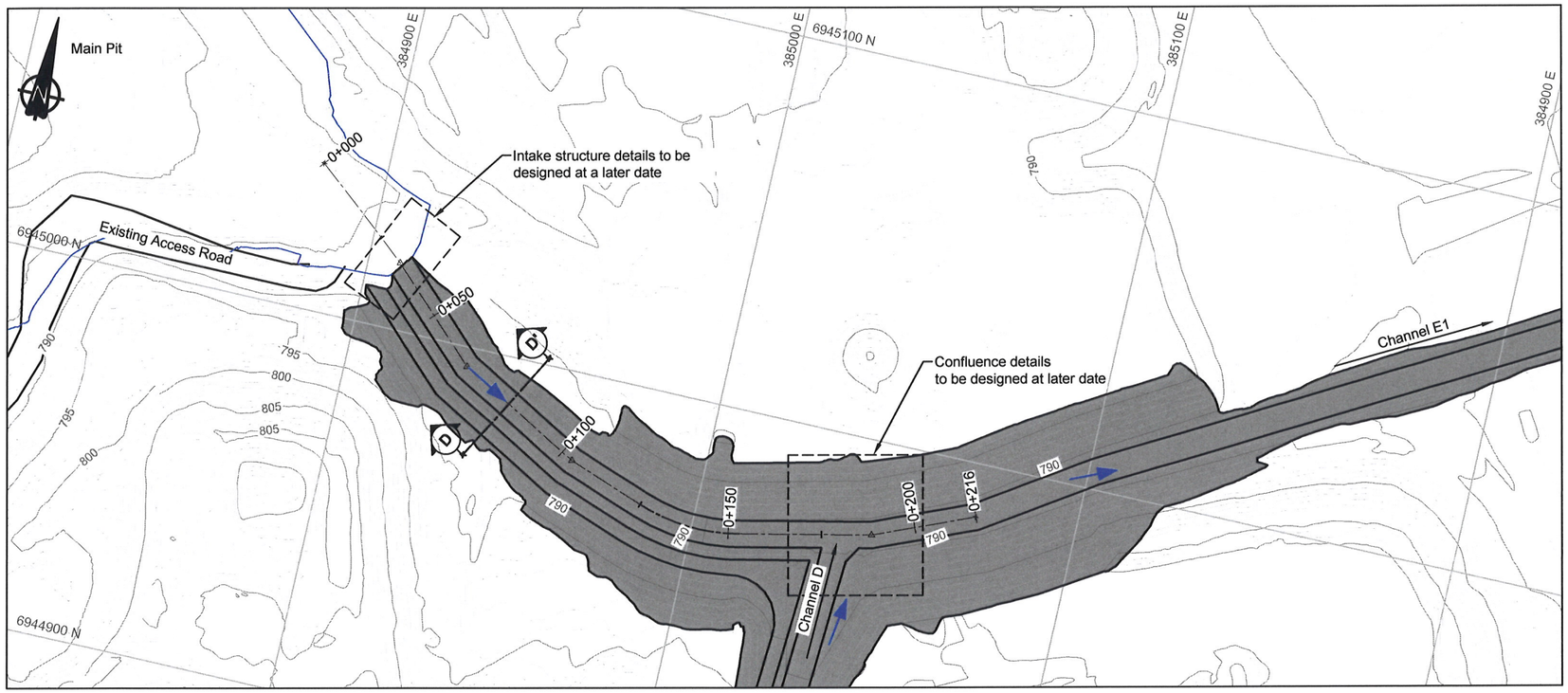
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Closure Conveyance Design

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LEGEND

- Conveyance Channel Design
- Channel Fill
- Channel Excavation
- Water Elevation
- Design Toes & Crests
- Flow Direction

NOTES

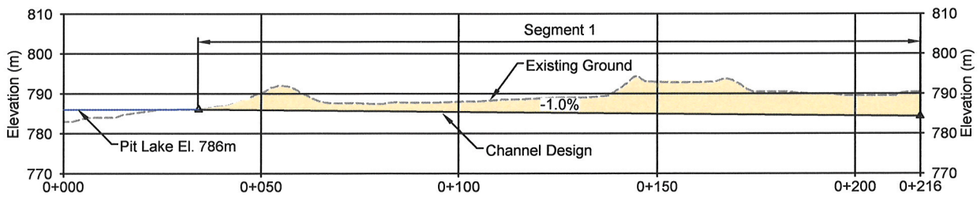
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2. Digital topographic surfaces have been provided by Minto Explorations Ltd.
3. The base of the channel will be graded to create a thalweg in the centerline.

PLAN VIEW CHANNEL C

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Scale in Metres

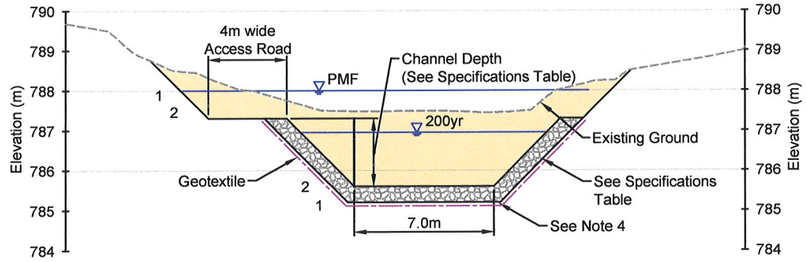
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Segment	Length (m)	Slope (%)	Minimum Channel Depth (m)	Channel Bottom Width (m)	Water Depth for 1:200 yrs (m)	Rip-rap Class (kg)	Rip-rap Thickness (mm)	D_{50} (mm)
1	182	-1.0	1.70	7.00	1.35	10	400	200



PROFILE CHANNEL C

0 5 10 15 20 25
Scale in Metres



SECTION D TYPICAL SECTION

0 1 2 3 4 5
Horizontal Scale in Metres
Vertical Exaggeration 2X

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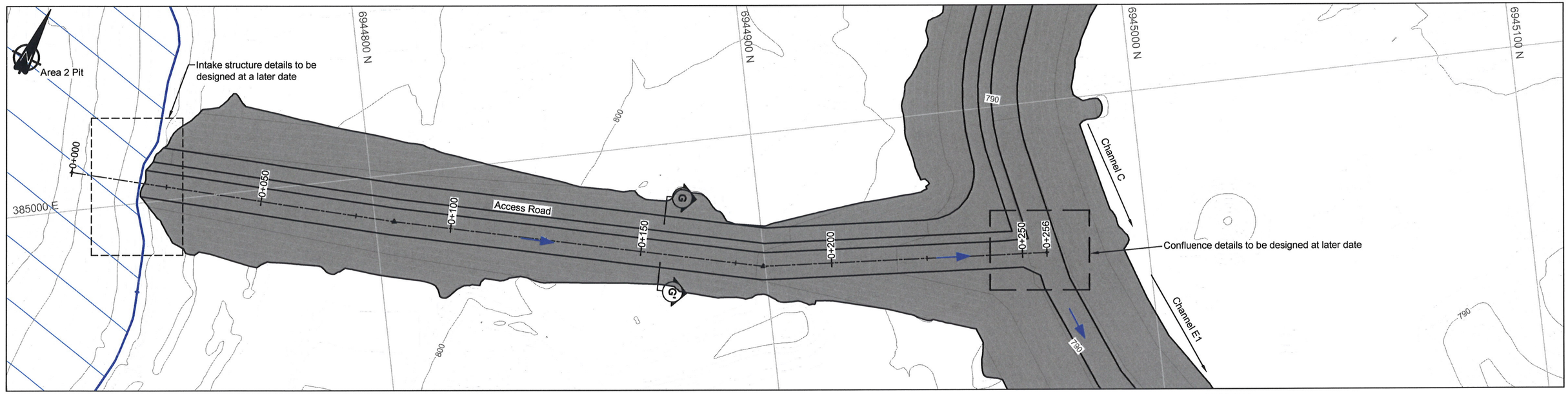
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Closure Conveyance Design

DRAWING TITLE: Channel C

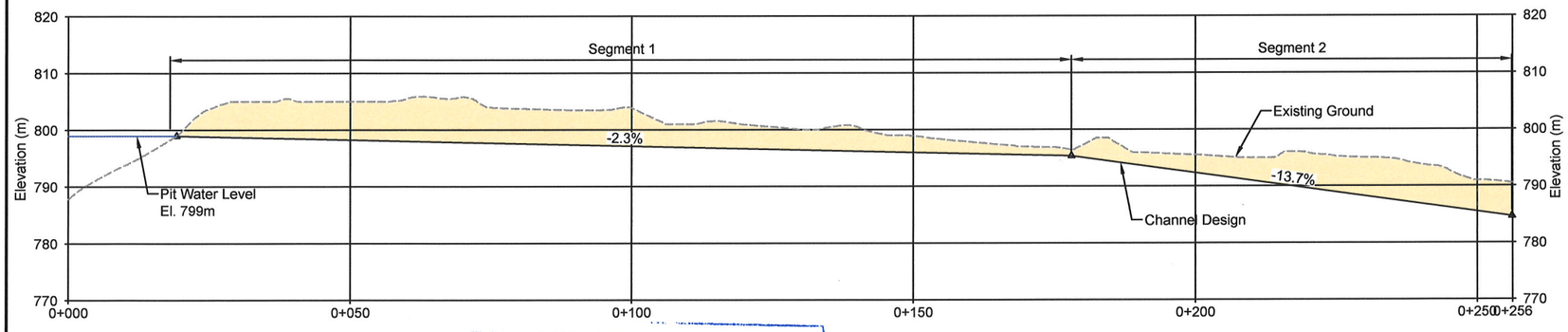
DRAWING NO.	SHEET	REVISION NO.
CC-04	5 OF 12	B

P:\01_SITES\Minto Mine YK Capstone\1040.CAD\IFR - Conveyance and Wetlands\1CM002_042-1.dwg



PLAN VIEW CHANNEL D

0 5 10 15 20 25
Scale in Metres



PROFILE CHANNEL D

0 5 10 15 20 25
Scale in Metres

CHANNEL D SPECIFICATIONS:

Segment	Length (m)	Slope (%)	Minimum Channel Depth (m)	Channel Bottom Width (m)	Water Depth for 1:200 yrs (m)	Rip-rap Class (kg)	Rip-rap Thickness (mm)	D_{50} (mm)
1	160	-2.3	2.10	7.00	1.80	10	400	200
2	78	-13.70	1.40	7.00	1.10	100	900	450

PERMIT TO PRACTICE
Signature
Date 2/06/2018
PERMIT NUMBER: PP019
Association of Professional Engineers of Yukon

PROFESSIONAL ENGINEER YUKON TERRITORY VICTOR ALEJANDRO MUNOZ SAAVEDRA 2/05/2018

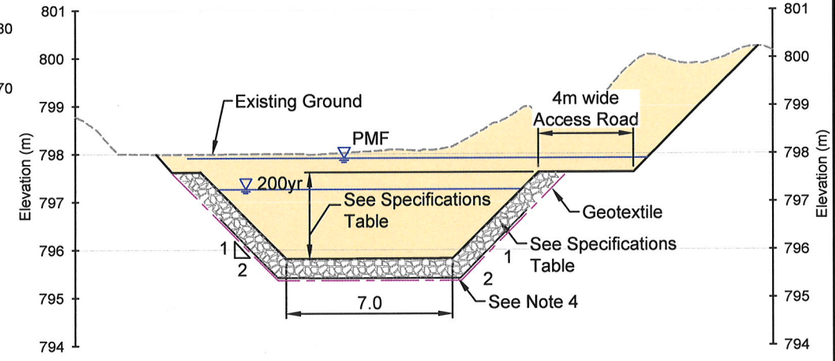
NOT FOR CONSTRUCTION

LEGEND

- Conveyance Channel Design
- Channel Fill
- Channel Excavation
- Pit Lake Elevation
- Design Toes & Crests
- Flow Direction

NOTES

1. Contour interval 1m.
2. Digital topographic surfaces have been provided by Minto Explorations Ltd.
3. Access road paralleling channel to be connected with the site wide access roads for mine closure.
4. Rip Rap classification gradation tables are provided in Drawing CC-06.
5. The base of the channel will be graded to create a thalweg in the centerline.



G SECTION G

0m 1 2 3 4 5
Horizontal Scale in Metres
Vertical Exaggeration 2X

DRAWING NO.	DRAWING TITLE	DRAWING NO.	DRAWING TITLE

REFERENCE DRAWINGS

NO.	DESCRIPTION	CHKD	APPD	DATE
B	Issued for Review	SP	VM	16Jan18
A	Issued for Review	SM	VM	15Jul16

REVISIONS

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SM	NV/TH	SM
CHECKED:	APPROVED:	DATE:
SM	VM	2018/01/16

FILE NAME: 1CM002_042-1.dwg

capstone MINTO MINE OPERATED BY MINTO EXPLORATIONS LTD.

MINTO EXPLORATIONS Ltd.

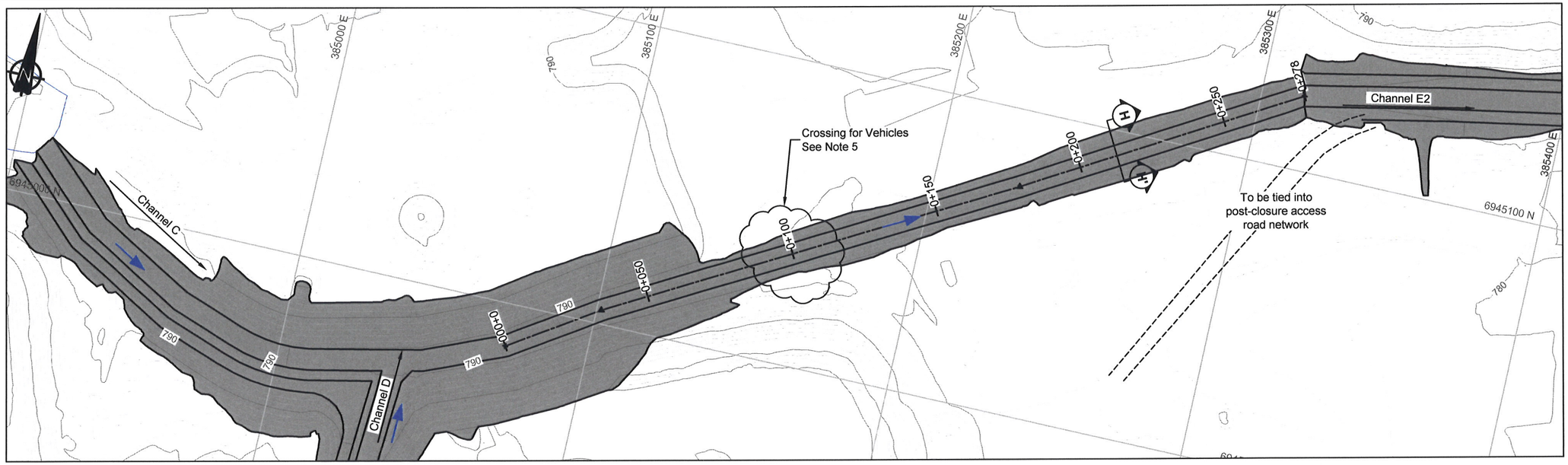
SRK JOB NO.: 1CM002.042

Closure Conveyance Design

DRAWING TITLE: Channel D

DRAWING NO.	SHEET	REVISION NO.
CC-05	6 OF 12	B

P:\01_SITES\Minto Mine YK-Capstone\1M0\CAD\FR - Conveyance and Wetlands\1CM002_042-1.dwg



LEGEND

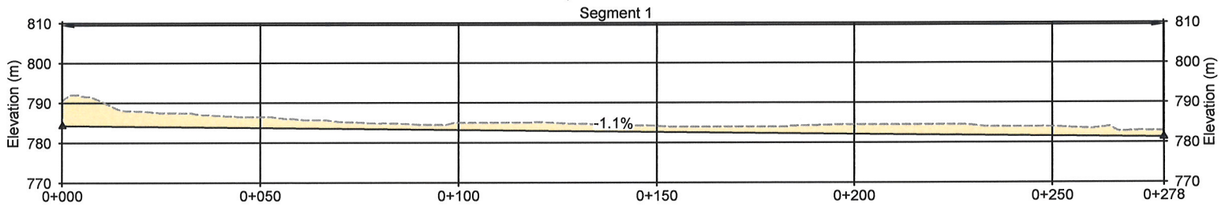
- Conveyance Channel Design
- Channel Excavation
- Design Toes & Crests
- Flow Direction

NOTES

1. Contour interval 1m.
2. Digital topographic surfaces have been provided by Minto Explorations Ltd.
3. Channel E is constructed as a typical channel. The base of the channel will be graded to create a thalweg in the centerline.
4. A light vehicle crossing is to be designed into the closure conveyance channel upon the completion of the channel construction. A swale crossing with 5:1 side slopes should be considered during design and alignment selection.
5. Surrounding area along Channel E1 graded to create a valley bottom. PMF will be conveyed downstream to Channel E2.
6. Segment 1 and Segment 3 set to a minimum freeboard of 10cm because of topographical constraint.

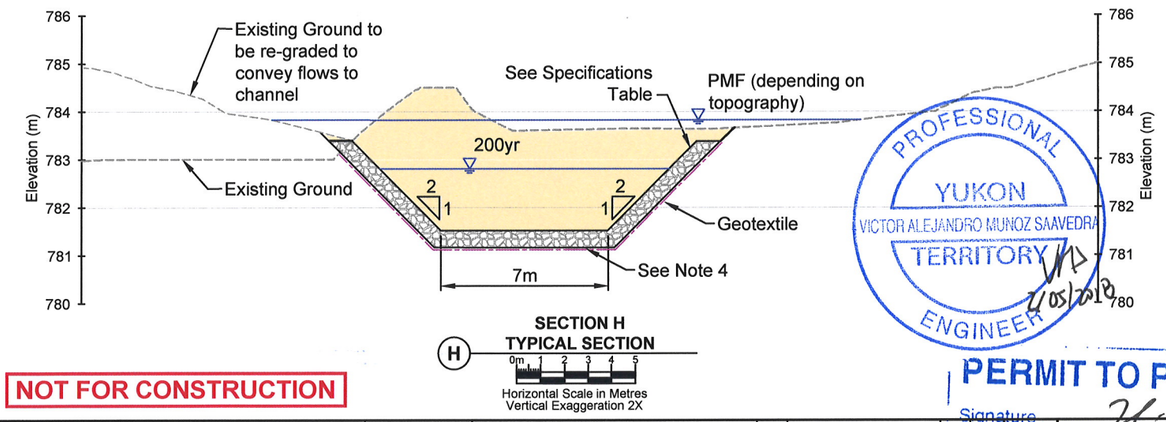
PLAN VIEW CHANNEL E1

0 5 10 15 20 25
Scale in Metres



PROFILE CHANNEL E1

0 5 10 15 20 25
Scale in Metres



SECTION H
TYPICAL SECTION

Horizontal Scale in Metres
Vertical Exaggeration 2X

10 kg CLASS - RIP RAP

Material Type	Rip Rap Dia. (mm)
D_{100}	300
D_{85}	280
D_{50}	200
D_{15}	90
Layer Thickness	400

25kg CLASS - RIP RAP

Material Type	Rip Rap Dia. (mm)
D_{100}	450
D_{85}	380
D_{50}	300
D_{15}	130
Layer Thickness	600

100 kg CLASS - RIP RAP

Material Type	Rip Rap Dia. (mm)
D_{100}	680
D_{85}	610
D_{50}	450
D_{15}	200
Layer Thickness	900

250kg CLASS - RIP RAP

Material Type	Rip Rap Dia. (mm)
D_{100}	900
D_{85}	820
D_{50}	600
D_{15}	270
Layer Thickness	1200

500kg CLASS - RIP RAP

Material Type	Rip Rap Dia. (mm)
D_{100}	1200
D_{85}	1030
D_{50}	800
D_{15}	340
Layer Thickness	1600

1000kg CLASS - RIP RAP

Material Type	Rip Rap Dia. (mm)
D_{100}	1500
D_{85}	1300
D_{50}	1000
D_{15}	420
Layer Thickness	2000

CHANNEL E SPECIFICATIONS:

Segment	Length (m)	Slope (%)	Minimum Channel Depth (m)	Channel Bottom Width (m)	Water Depth for 1:200 yrs (m)	Rip-rap Class (kg)	Rip-rap Thickness (mm)	D_{50} (mm)
1	278	-1.1	2.0	7.0	1.82	10	400	450
2	260	-6.2	2.0	7.0	1.26	100	900	450
3	98	-1.4	2.0	7.0	1.89	100	900	450
4	92	-5.8	2.0	7.0	1.28	100	900	450
5	92	-6.3	2.0	7.0	1.26	100	900	450
6	105	-4.1	2.0	7.0	1.41	100	900	450
7	33	-7.6	2.0	7.0	1.19	250	1200	600
8	46	-5.1	2.0	7.0	1.33	100	900	450
9	150	-20.7	2.0	7.0	0.90	1000	2000	1000
10	40	-0.05	3.3	20.0	1.50	100	900	300

PROFESSIONAL ENGINEER
YUKON TERRITORY
VICTOR ALEJANDRO MUNOZ SAAVEDRA
4/05/2018

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Date 2/06/2018
PERMIT NUMBER: PP019
Association of Yukon Professional Engineers
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DRAWING NO.	DRAWING TITLE	DRAWING NO.	DRAWING TITLE
REFERENCE DRAWINGS			

NO.	DESCRIPTION	CHKD	APPD	DATE
B	Issued for Review	SP	VM	16Jan18
A	Issued for Review	SM	VM	18Jan18
REVISIONS				

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CHECKED:	SM	APPROVED:	VM	DATE:	2018/01/16
FILE NAME:	1CM002_042-1.dwg				

capstone MINTO MINE
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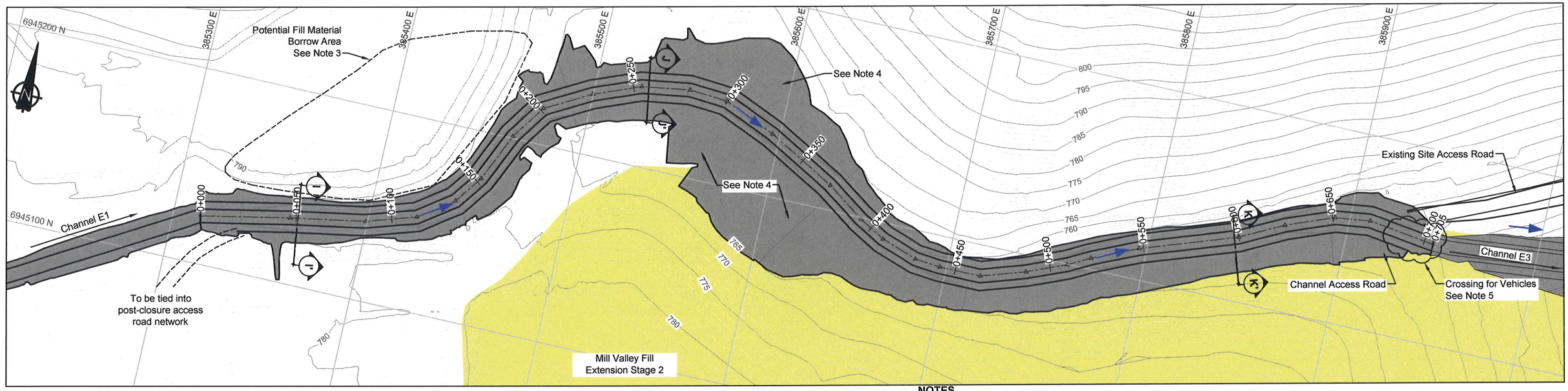
SRK JOB NO.: 1CM002.042

Closure Conveyance Design

DRAWING TITLE: Channel E1

DRAWING NO.	SHEET	REVISION NO.
CC-06	7 OF 12	B

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LEGEND

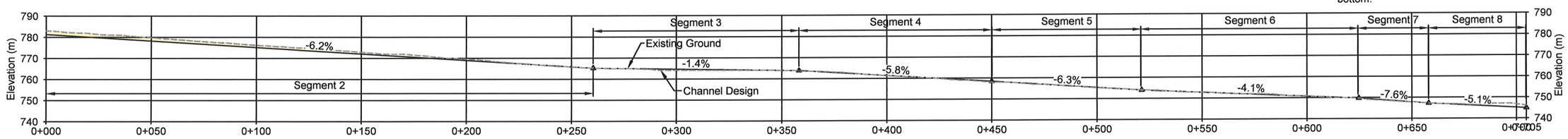
- Conveyance Channel Design
- Phase VIVI Infrastructure (Existing and Planned)
- Channel Excavation
- Design Toes & Crests
- Flow Direction

PLAN VIEW CHANNEL E2

0 10 20 30 40 50
Scale in Metres

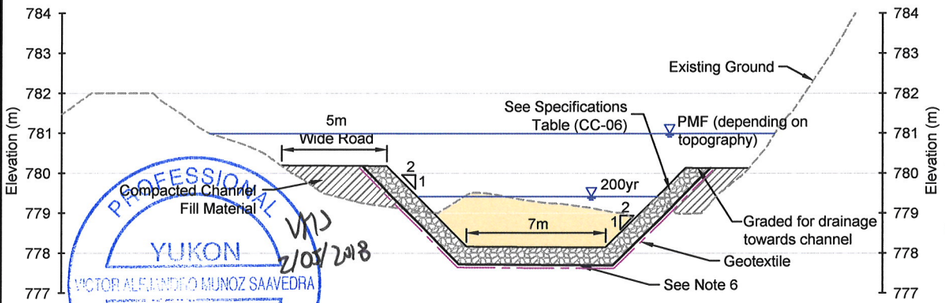
NOTES

1. Contour interval 1m.
2. Digital topographic surfaces have been provided by Minto Explorations Ltd.
3. Extents and depth of the potential fill material borrow area has not be defined for this stage of design. Area identified as a potential source due to overburden / fill thickness.
4. Backfill and regrade required to convey surface water to conveyance channel
5. A light vehicle crossing is to be designed into the closure conveyance channel upon the completion of channel construction. A swale crossing with 5:1 side slopes should be considered during design and alignment selection.
6. The base of the channel will be graded to create a thalweg in the centerline
7. Surrounding area along Channel E2 to be graded to create a valley bottom.



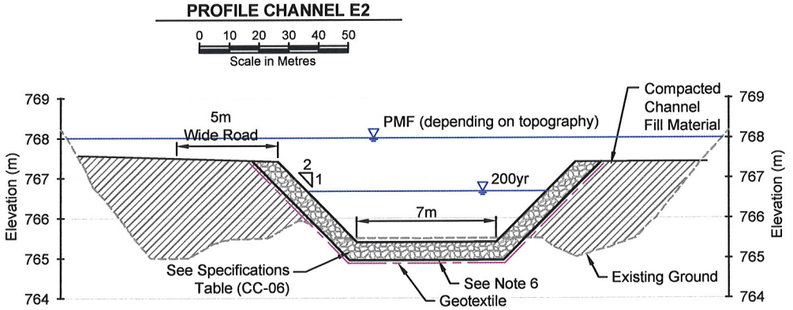
PROFILE CHANNEL E2

0 10 20 30 40 50
Scale in Metres



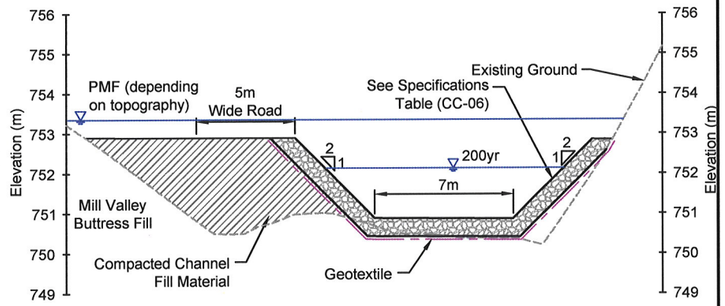
SECTION I

0 2 4 6 8 10
Horizontal Scale in Metres
Vertical Exaggeration 2X



SECTION J

0 2 4 6 8 10
Horizontal Scale in Metres
Vertical Exaggeration 2X



SECTION K

0 2 4 6 8 10
Horizontal Scale in Metres
Vertical Exaggeration 2X

PROFESSIONAL ENGINEER YUKON TERRITORY VICTOR ALEJANDRO MUNOZ SAAVEDRA 2/06/2018

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Signature
Date 2/06/2018
PERMIT NUMBER: PP019
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DRAWING NO.	DRAWING TITLE	DRAWING NO.	DRAWING TITLE
REFERENCE DRAWINGS			

NO.	DESCRIPTION	CHK'D	APP'D	DATE
B	Issued for Review	SM	VM	16Jan18
A	Issued for Review	SM	VM	15Jul16

REVISIONS

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DESIGN: SM	DRAWN: NV/TH	REVIEWED: SM
CHECKED: SM	APPROVED: VM	DATE: 2018/01/16

FILE NAME: 1CM002_042-1.dwg

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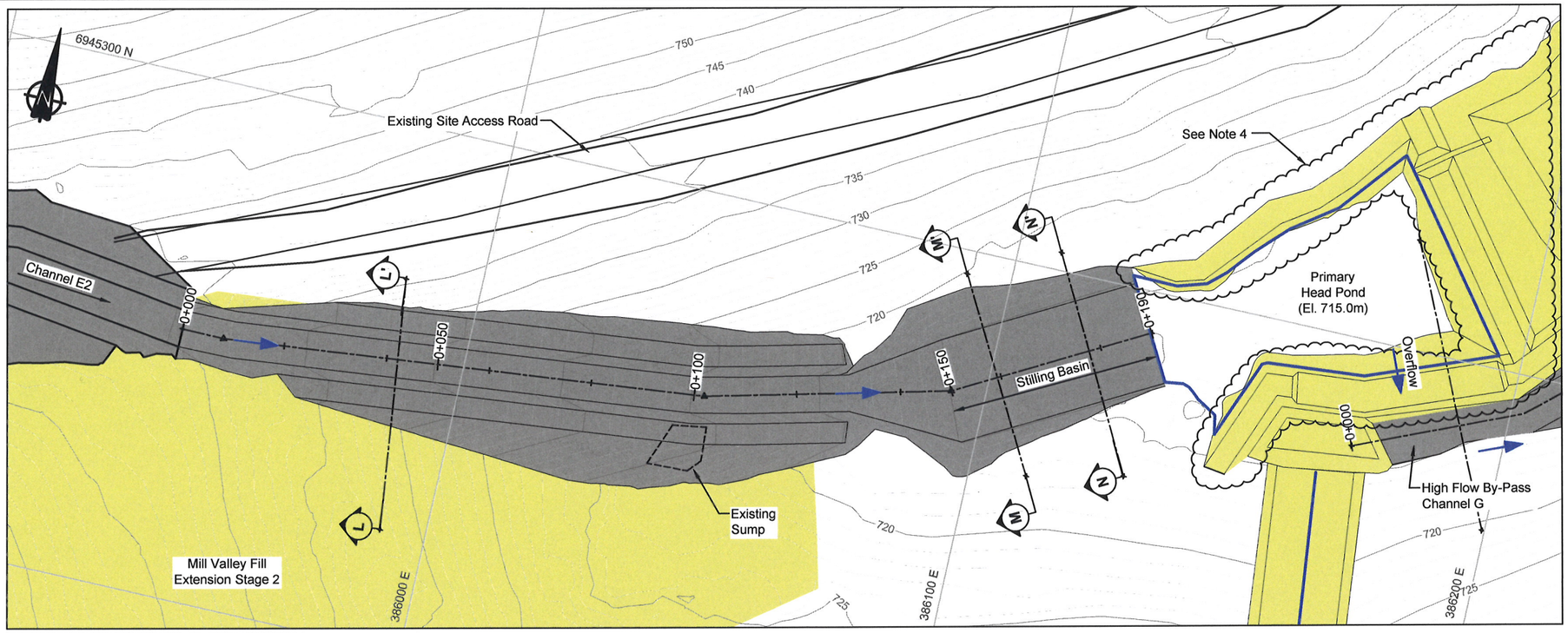
SRK JOB NO.: 1CM002.042

Closure Conveyance Design

DRAWING TITLE: Channel E2

DRAWING NO.	SHEET	REVISION NO.
CC-07	8 OF 12	B

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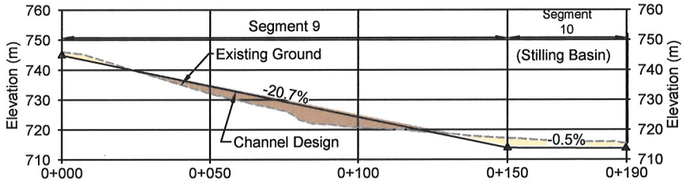


LEGEND

- Conveyance Channel Design
- Channel Foundation Fill
- Channel Excavation
- Phase VIVI Infrastructure (Existing and Planned)
- Design Toes & Crests
- Flow Direction

NOTES

1. Contour interval 1m.
2. Digital topographic surfaces have been provided by Minto Explorations Ltd.
3. Stilling basin design rip rap specifications are located in the specifications tables in Drawing CC-06.
4. Wetland Infrastructure Design prepared by SRK. Refer to Wetland Infrastructure Design Report (SRK 2016).
5. The base of the channel will be graded to create a thalweg in the centerline
6. Surrounding area along Channel E3 graded to create a valley bottom. PMF will be conveyed downstream to the stilling basin.

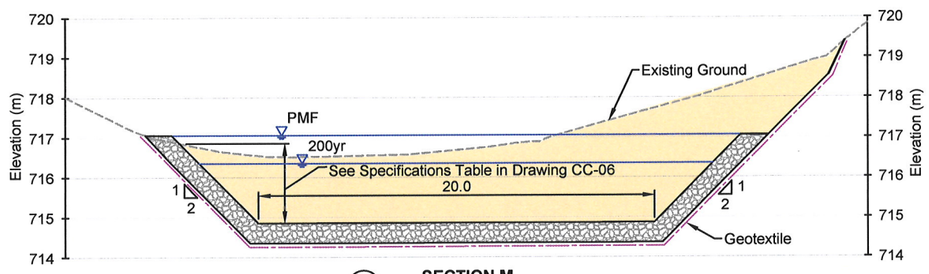


PROFILE CHANNEL E3

0 10 20 30 40 50
Scale in Metres

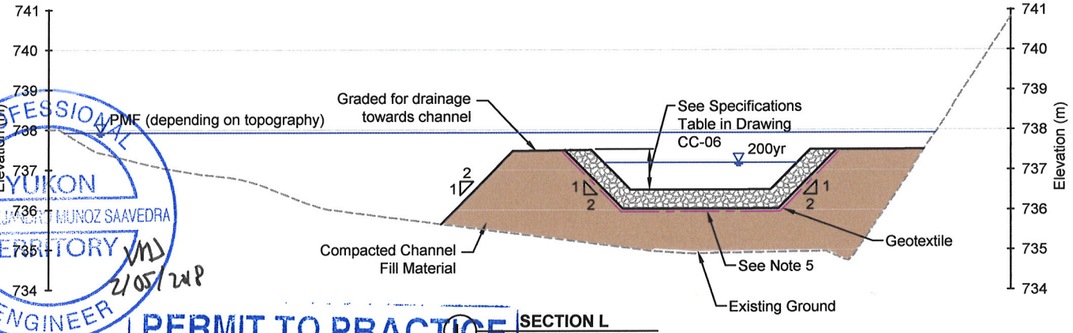
PLAN VIEW CHANNEL E3

0 5 10 15 20 25
Scale in Metres



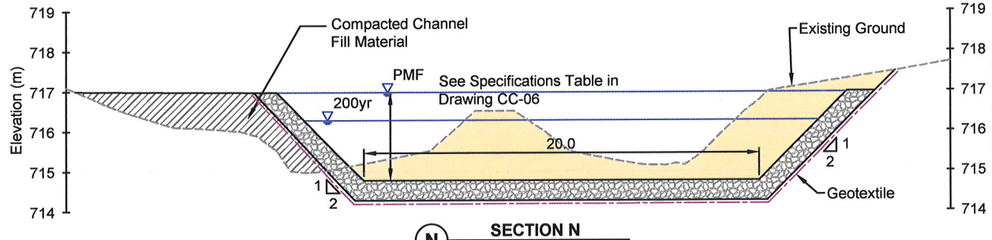
M SECTION M

0 1 2 3 4 5
Horizontal Scale in Metres
Vertical Exaggeration 2X



L SECTION L

0 1 2 3 4 5
Horizontal Scale in Metres
Vertical Exaggeration 2X



N SECTION N

0 1 2 3 4 5
Horizontal Scale in Metres
Vertical Exaggeration 2X

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PROFESSIONAL ENGINEER YUKON TERRITORY VICTOR ALEJANDRO MUNOZ SAAVEDRA 2/05/2018

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Signature
Date 2/06/2018
PERMIT NUMBER: PP019
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REFERENCE DRAWINGS

NO.	DESCRIPTION	CHK'D	APP'D	DATE
B	Issued for Review	SP	VM	16Jan18
A	Issued for Review	SM	VM	15Jul16

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SM	VM	2018/01/16

FILE NAME: 1CM002_042-1.dwg

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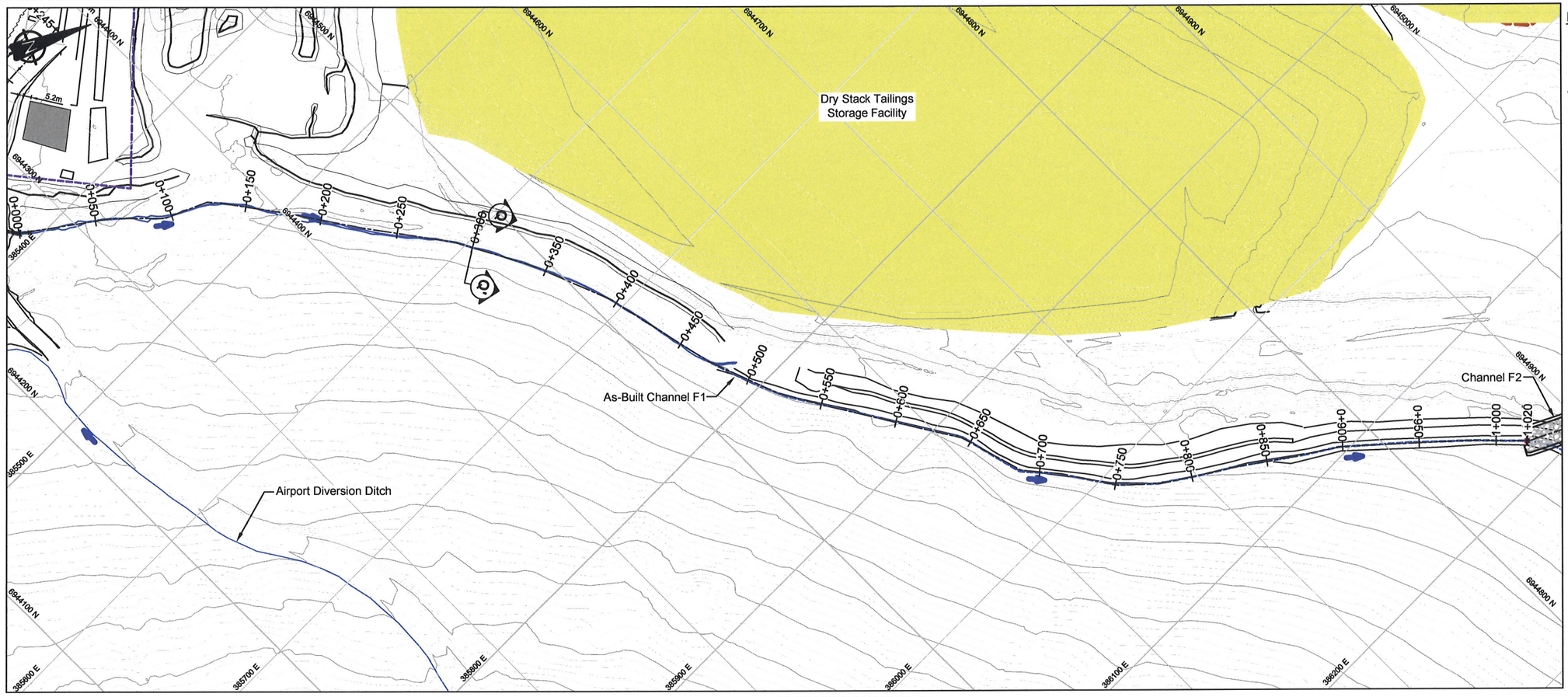
SRK JOB NO.: 1CM002.042

Closure Conveyance Design

DRAWING TITLE:
Channel E3
and Stilling Basin

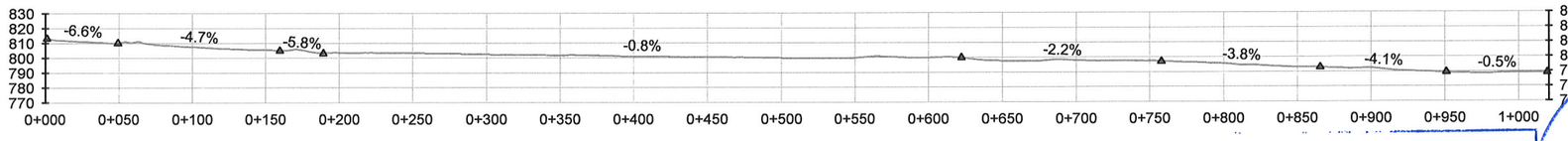
DRAWING NO.	SHEET	REVISION NO.
CC-08	9 OF 12	B

P:\01_SITES\Minto Mine YK Capstone\040 CAD\FR - Conveyance and Wetlands\1CM002_042-1.dwg



PLAN VIEW CHANNEL F1

0m 10 20 30 40 50 60



AS-BUILT PROFILE CHANNEL F1

0m 20 40 60 80

LEGEND

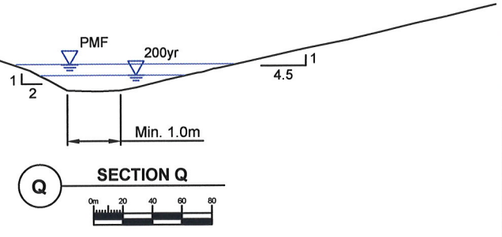
- Existing Major Contour
- Existing Minor Contour
- Diversion Channel (existing)
- Diversion Channel (design by others)
- Excavation Limits
- Water Flow Direction
- Phase VIVI Infrastructure (Existing and Planned)

NOTES

1. Existing topographic contours provided by Minto dated August 2017.
2. Contour interval is 1m.

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PERMIT NUMBER: PP019
Association of Professional Engineers of Yukon

PROFESSIONAL ENGINEER YUKON TERRITORY ENGINEER
2/05/2018



SECTION Q

0m 20 40 60 80

NOT FOR CONSTRUCTION

DRAWING NO.	DRAWING TITLE	DRAWING NO.	DRAWING TITLE
-	-		

REFERENCE DRAWINGS

NO.	DESCRIPTION	CHK'D	APP'D	DATE
F	Issued for Review	SP	VM	16Jan18
E	Issued for Discussion	-	-	26Aug14
D	Issued for Discussion	-	-	28Jul14
C	Issued for Discussion	-	-	1Nov13
B	Issued for Discussion	-	-	7Aug13
A	Issued for Review	-	-	6Jun13

REVISIONS

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DESIGN:	DRAWN:	REVIEWED:
VM	DJ / TH	PM
CHECKED: -	APPROVED: -	DATE: 2017/01/16

FILE NAME: 1CM002_042-Diversions-Rev_E.dwg

capstone MINTO MINE
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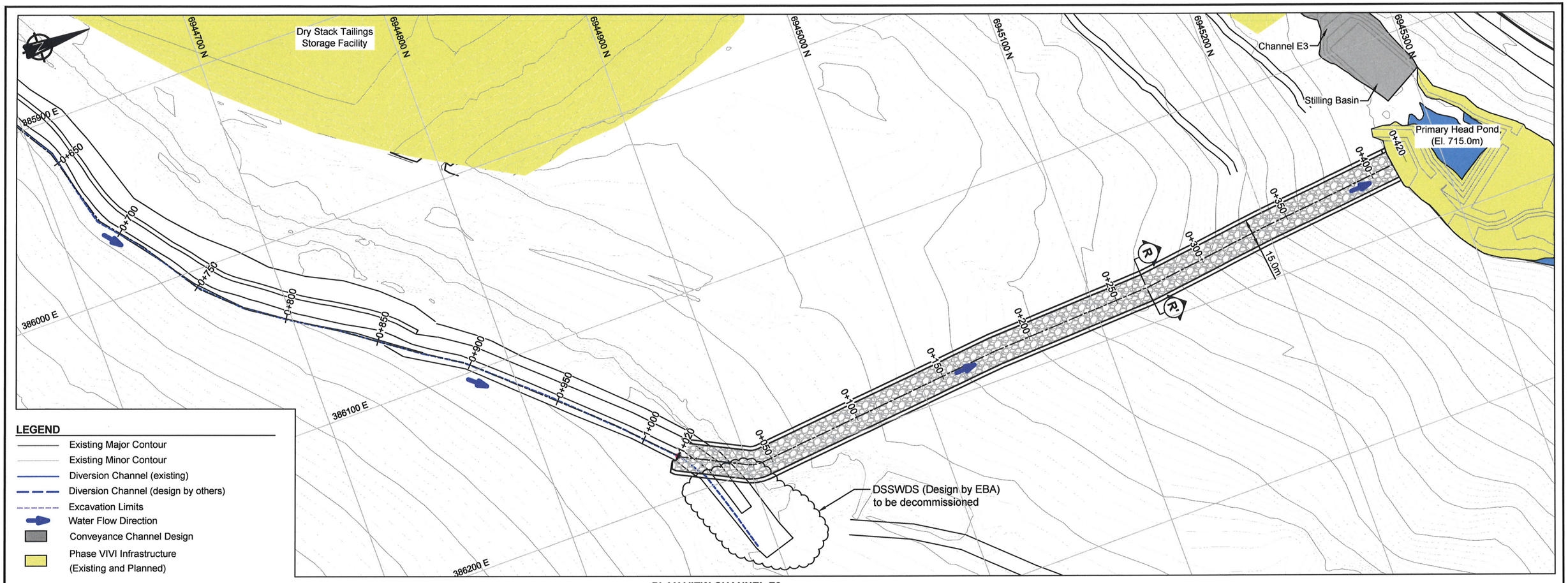
SRK JOB NO.: 1CM002.029

Closure Conveyance Design

DRAWING TITLE:
Channel F1
Existing Tailings Diversion Ditch

DRAWING NO.	SHEET	REVISION NO.
CC-09	10 of 12	F

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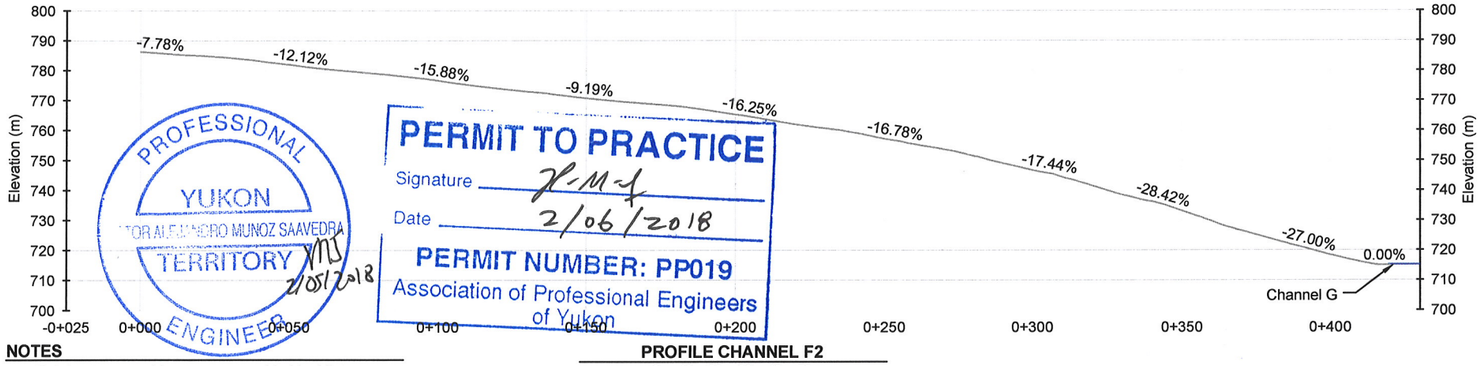


LEGEND

- Existing Major Contour
- Existing Minor Contour
- Diversion Channel (existing)
- Diversion Channel (design by others)
- Excavation Limits
- Water Flow Direction
- Conveyance Channel Design
- Phase VIVI Infrastructure (Existing and Planned)

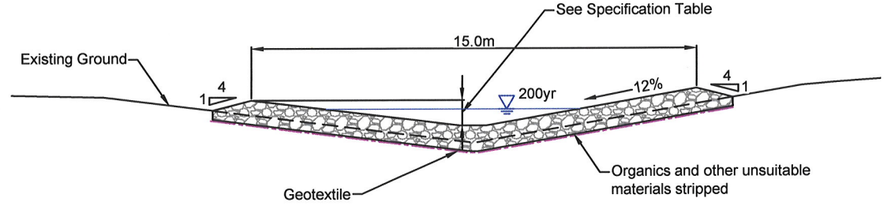
PLAN VIEW CHANNEL F2

0 10 20 30 40 50
Scale in Metres



PROFILE CHANNEL F2

0 10 20 30 40 50
Scale in Metres



SECTION R

0 10 20 30 40 50
Scale in Metres

CHANNEL F SPECIFICATIONS:

Length (m)	Minimum Channel Depth (m)	Rip-rap Class (kg)	Rip-rap Thickness (mm)	D_{50} (mm)
420	0.8	500	1600	800

NOTES

1. Existing topographic contours provided by Minto dated January 2014.
2. Contour interval is 1m.
3. Rip-rap sized for steepest grade (28.42%)

PROFESSIONAL ENGINEER YUKON VICTOR ALEJANDRO MUNOZ SAAVEDRA TERRITORY VMS 2/05/2018

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Signature
Date 2/06/2018
PERMIT NUMBER: PP019
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DRAWING NO.	DRAWING TITLE	DRAWING NO.	DRAWING TITLE
-	-		

REFERENCE DRAWINGS

NO.	DESCRIPTION	CHK'D	APP'D	DATE
F	Issued for Review	SP	VM	16Jan18
E	Issued for Discussion	-	-	26Aug14
D	Issued for Discussion	-	-	28Jul14
C	Issued for Discussion	-	-	1Nov13
B	Issued for Discussion	-	-	7Aug13
A	Issued for Review	-	-	6Jun13

REVISIONS

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srk consulting

DESIGN:	DRAWN:	REVIEWED:
VM	DJ / TH	PM
CHECKED: -	APPROVED: -	DATE: 2018/01/16

FILE NAME: 1CM002_042-Diversions-Rev_E.dwg

capstone MINTO MINE
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Minto Explorations Ltd.

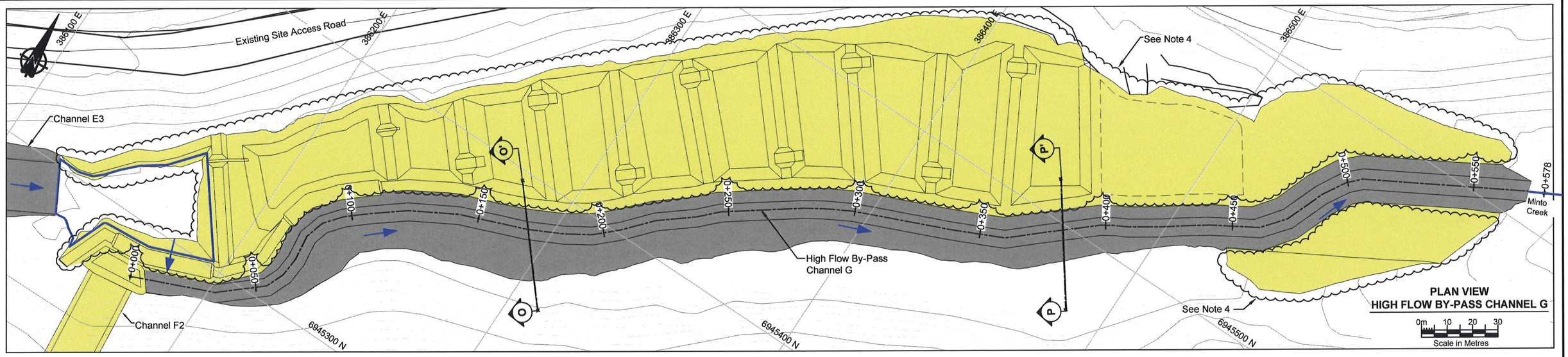
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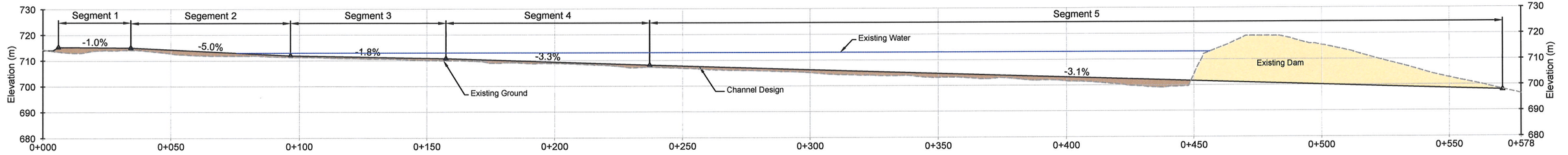
Closure Conveyance Design

DRAWING TITLE: Channel F2

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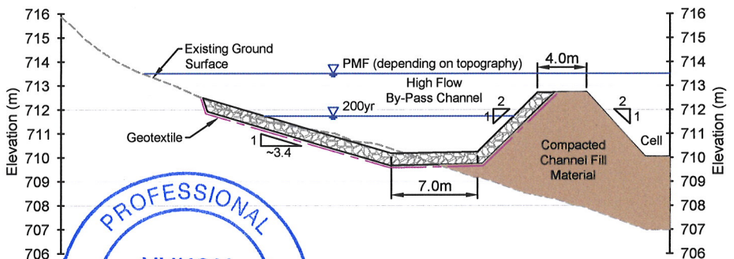
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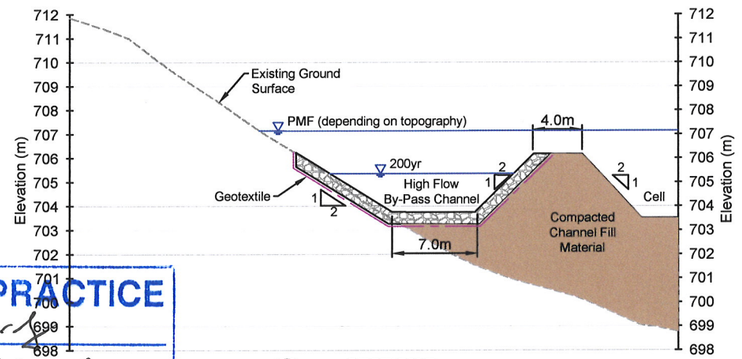




CENTRELINE PROFILE
HIGH FLOW BY-PASS CHANNEL G

0m 10 20 30
Scale in Metres





O — SECTION O
0 2 4 6 8 10
Scale in Metres

P — SECTION P
0 2 4 6 8 10
Scale in Metres
Horizontal Scale in Metres
Vertical Exaggeration 2X

PROFESSIONAL ENGINEER
YUKON TERRITORY
VICTOR ALEJANDRO MUNOZ SAAVEDRA
2/05/2018

PERMIT TO PRACTICE
Signature
Date 2/06/2018
PERMIT NUMBER: PP019
Association of Professional Engineers of Yukon

NOT FOR CONSTRUCTION

LEGEND

- Conveyance Channel Design
- Channel Foundation Fill
- Channel Excavation
- Designed Infrastructure (Not Constructed)
- Design Toes & Crests

NOTES

1. Contour interval 1m.
2. Digital topographic surfaces have been provided by Minto Explorations Ltd.
3. Wetland Infrastructure Design prepared by SRK. Refer to Wetland Infrastructure Design Report (SRK 2016).
4. 200 year channel depth along channel varies depending on topography.

CHANNEL HIGH FLOW BY-PASS CHANNEL SPECIFICATIONS:

Segment	Length (m)	Slope (%)	Minimum Channel Depth (m)	Channel Bottom Width (m)	Rip-rap Class (kg)	Rip-rap Thickness (mm)	D_{50} (mm)
1	28	-1.0	2.5	7.0	25	600	300
2	62	-5.0	2.0	7.0	25	600	300
3	61	-1.8	2.0	7.0	25	600	300
4	84	-3.3	2.0	7.0	25	600	300
5	334	-3.1	2.0	7.0	25	600	300

REFERENCE DRAWINGS				REVISIONS					
DRAWING NO.	DRAWING TITLE	DRAWING NO.	DRAWING TITLE	NO.	DESCRIPTION	CHK'D	APP'D	DATE
				B	Issued for Review	SP	VM	16Jan18
				A	Issued for Review	SM	VM	15Jul16

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capstone MINTO MINE
OPERATED BY MINTO EXPLORATIONS LTD.

MINTO EXPLORATIONS Ltd.

SRK JOB NO.: 1CM002.042

Closure Conveyance Design

DRAWING TITLE:
High Flow By-Pass Channel G

DRAWING NO.	SHEET	REVISION NO.
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Appendix B – Site-Specific Hydrology Review

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Memo

To:	Ryan Herbert, Minto Explorations Ltd.	**Client:**	Minto Explorations Ltd.
From:	Samantha Barnes Victor Muñoz S. Francis Smith	**Project No:**	1CM002.042
Cc:	Dylan MacGregor, SRK	**Date:**	August 3, 2016
Subject:	Site-Specific Hydrology Review, Minto Mine, Yukon		

1 Introduction

SRK Consulting (Canada) Inc. has prepared an updated closure water conveyance plan design report for Minto Mine (Minto) located in Yukon, Canada, and operated by Minto Explorations Ltd. In support of the updates to the conveyance designs, SRK reviewed and updated site-wide hydrology models to include potential influences resulting from future climate change and alterations to watershed catchment areas. Inputs to the modelling are listed as follows:

- Rainfall depths for short duration storms for return periods ranging from the 1:2 through the 1:200 year precipitation event
- Maximum snowmelt depth expected over a 24-hour period
- Probable maximum precipitation (PMP) over a 24-hour period
- Hydrological curve number calibrated for regional catchment
- Hydrologic modelling approach to estimate peak flows for each structure

This memorandum outlines the formulation of inputs required for the closure conveyance designs.

2 Supporting Information from Public Databases

Regional historic precipitation data was obtained from the Meteorological Service of Canada (MSC) division of Environment Canada (EC 2015a). Regional stations were included in the studies for short duration rainfall, mean annual precipitation (MAP) and probable maximum precipitation (PMP).

Regional daily hydrometric data was obtained from the Water Survey of Canada (WSC) within an approximate 300 km radius surrounding the site for the unit peak flow analysis (EC 2015b).

Precipitation intensity–duration-frequency (IDF) estimation was derived utilizing short-duration rainfall IDF statistics obtained from the Government of Canada's Engineering Climate Dataset (Government of Canada 2016).

3 Precipitation

3.1 Mean Annual Precipitation

Regional regression analyses were performed using the 20 nearest MSC meteorological stations. RStudio® software (version 3.2.3; i386 - w64) was used to calculate mean annual precipitation for each station. Table 3-1 presents the station locations, distance relative to the site and their respective MAP estimates.

Table 3-1: Regional Meteorological Stations and MAP

Station Name	Longitude	Latitude	MAP [mm]	Elevation	Distance
PELLY RANCH	−137.4	62.8	309	454	23
CARMACKS	−136.3	62.1	282	525	75
STEWART CROSSING	−136.7	63.4	332	480	89
MCQUESTEN	−137.5	63.6	349	458	109
MAYO A	−135.9	63.6	304	504	131
STEWART RIVER	−139.4	63.3	294	358	136
DRURY CREEK	−134.4	62.2	373	609	154
SNAG A	−140.4	62.4	366	587	165
ELSA	−135.5	63.9	430	814	170
KENO HILL	−135.2	63.9	585	1473	178
DAWSON A	−139.1	64.0	323	370	184
BEAVER CREEK A	−140.9	62.4	413	649	188
BEAVER CREEK YTG	−140.9	62.4	414	663	189
ANVIL	−133.4	62.4	389	1158	200
FARO A	−133.4	62.2	318	716	205
KLONDIKE	−138.2	64.5	468	973	210
CLINTON CREEK	−140.7	64.5	344	576	269
OGILVIE RIVER	−138.3	65.4	340	597	310

Figure 3-1 includes a location map of the stations used in the MAP analysis. It also presents the subsequent regression analysis for MAP against elevation and the monthly precipitation distributions.

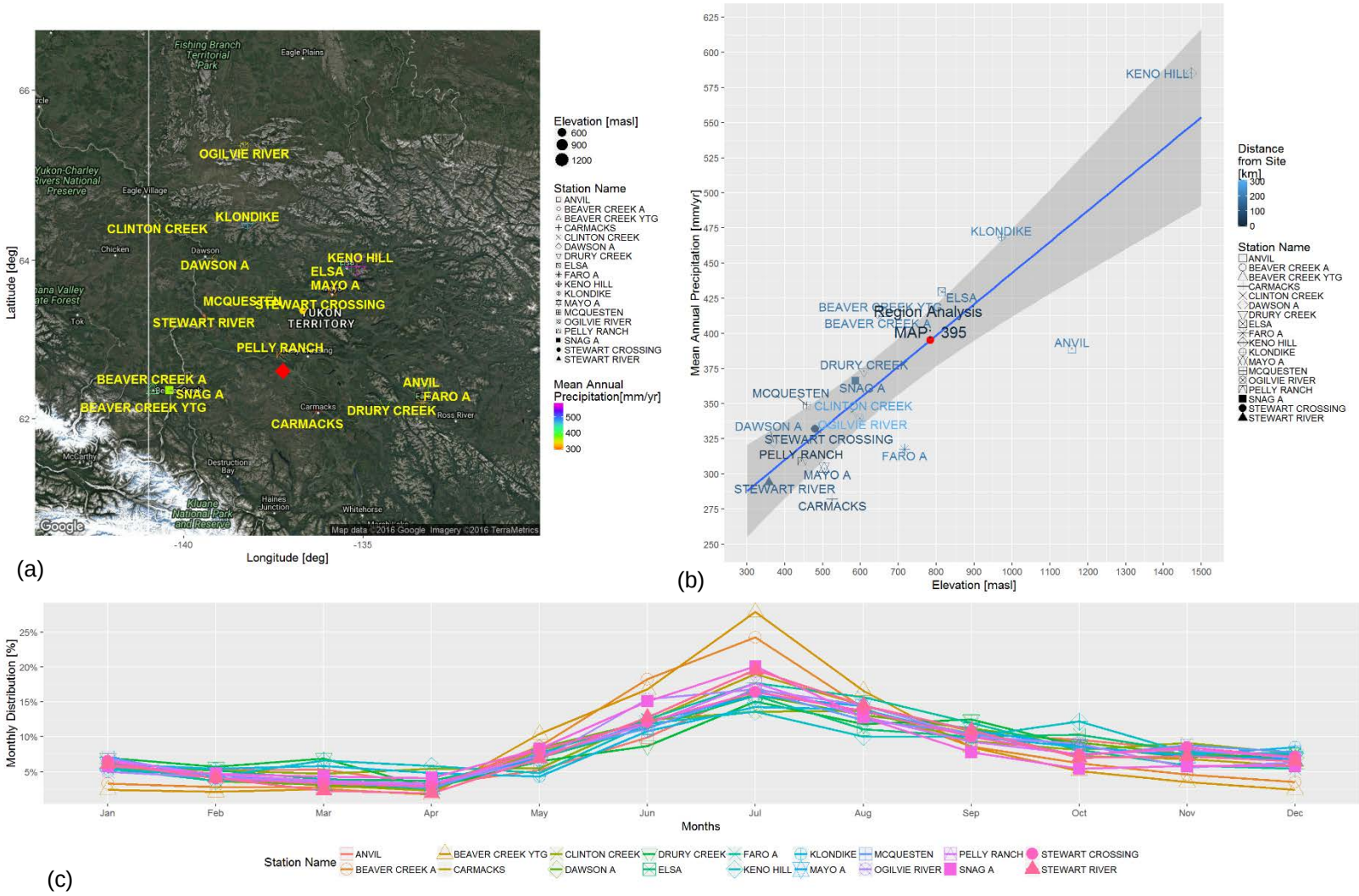


Figure 3-1: MAP Analysis Station Locations (a), Regional Regression Analysis (b), and Monthly Precipitation Distributions (c)

Based on the MAP versus elevation regression analysis and an average site elevation of 784 meters above sea level (masl), the MAP for Minto is estimated to be 395 mm.

3.2 Precipitation-Duration-Frequency Analysis

Short-duration storm analyses provide an estimation of the amount of precipitation expected for a storm return period ranging from 2 to 200 years with durations ranging from 5 minutes to 24 hours. A regional analysis was performed in order to understand the relationships between return period, duration, and rainfall for stations within close proximity to the site.

Using the results of the regional analysis, the precipitation-duration-frequency (PDF) data for the site was established using the following equation:

$$\text{Equation 1: } Pp_{Duration\ hrs.}^{Frequency\ years} = Pp_{24hrs.}^{10yrs} \cdot FF_{(Frequency)} \cdot DF_{(Duration)}$$

Where,

Pp is the precipitation as a function of storm frequency and duration;

$Pp_{24hrs.}^{10yrs}$ is the precipitation in 24 hours with a return period of 10 years;

FF are frequency factors, which correspond to the rate between precipitation associated with a defined frequency and the return period of 10 years; and

DF are duration factors, which correspond to the rate between the defined storm duration and a storm duration of 24 hours.

Duration and frequency factors for a 1:10 year, 24-hour precipitation event were established from the regional analysis. The following sections discuss how the design precipitation event was developed using the frequency factors.

3.2.1 One-in-Ten Year 24-hour Precipitation Event

This analysis was performed using the RStudio software, and was conducted using data obtained from the Environment Canada meteorological stations (Environment Canada 2015a) located throughout the Yukon (Figure 3-1). Frequency analyses were performed for each station to estimate the 1:10 year, 24-hour–duration precipitation depth. These values were then correlated with elevation, latitude, and longitude. The strongest correlation was with longitude, which was used to establish the 1:10 year precipitation for the site (Figure 3-2).

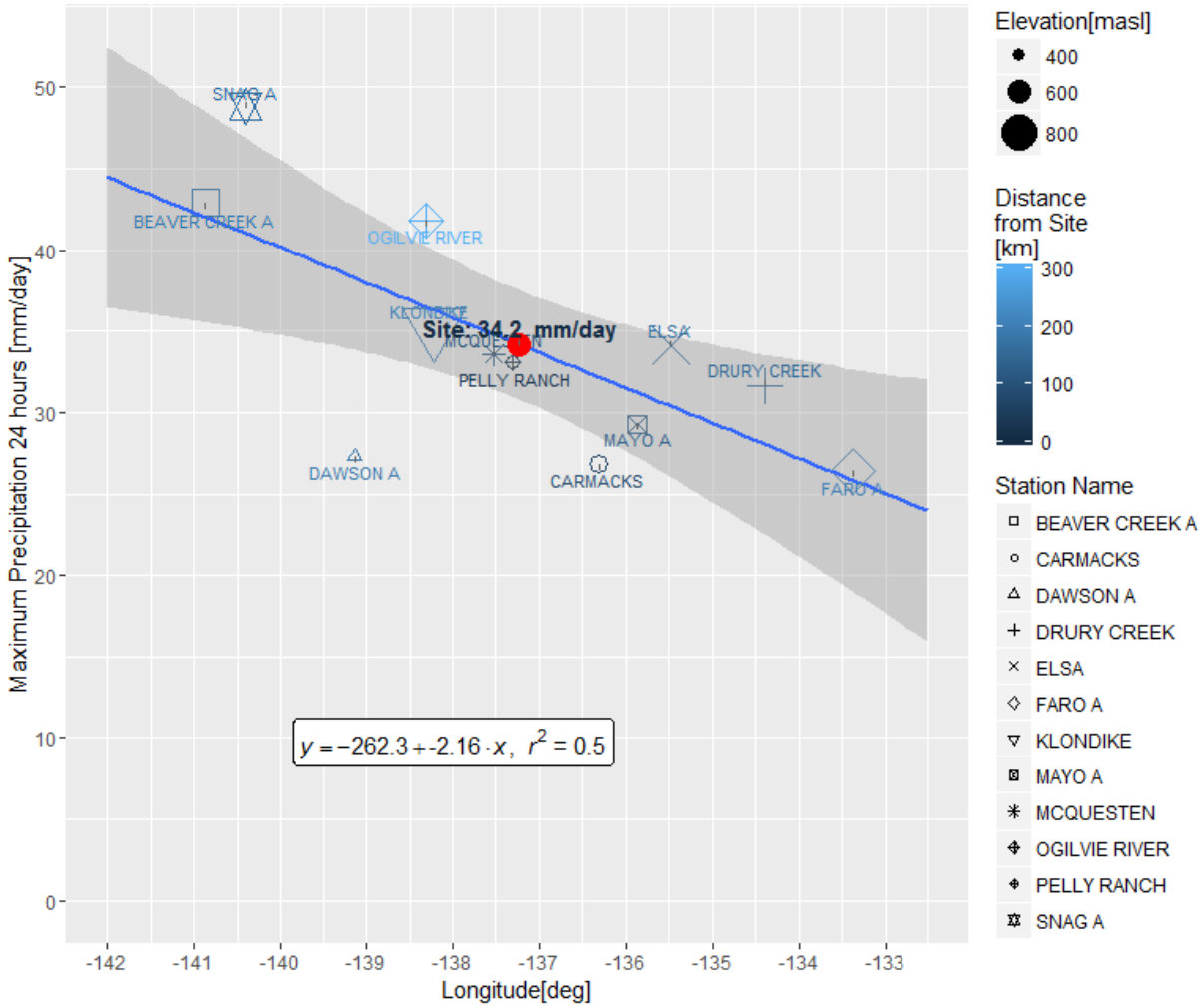


Figure 3-2: Regional Analysis for the 1:10 Year 24-hour Duration Rainfall Event

Based on the site longitude of −137.2 degrees, the 1:10 year 24-hour rainfall event is 34.2 mm.

3.2.2 Regional Intensity-Duration-Frequency Relationships

Five meteorological stations with IDF statistics all within 185 km of the site were located and are summarized in Table 3-2.

Table 3-2: Summary of Meteorological Stations Used for Short Duration Storm Event Analysis

Station Name	Latitude	Longitude	Elevation	Years of Data
Mayo A	−135.9	63.6	503.8	22
Pelly Ranch	−137.4	62.8	454.2	32
Dawson A	−139.1	64.0	370.3	26
Carmacks	−136.3	62.1	524.9	15
Burwash A	−139.1	61.4	806.2	13

The frequency analyses were prepared considering the following probabilistic distributions: Normal, Log Normal, GEV, Gumbel, Pearson III, and Log Pearson III. The selection of the distribution parameters was prepared with the L-moments methodology.

The L-moments approach provides a more robust estimation to outliers when compared with typical estimations. The selection of the best-fit distribution was prepared based on four criteria: Akaike Information Criterion, Corrected Akaike Information Criterion, Anderson-Darling Criterion, and Bayesian Information Criterion, as described by (Laio, Di Baldassarre, and Montanari 2009). These best-fit distributions were implemented using RStudio.

The results from the frequency analyses were compiled into an IDF table for each station. Duration and frequency factors for each regional station were calculated using an iterative optimization process, such that the product of the two factors and the 1:10 year, 24-hour precipitation depth would equal the rainfall depth for a given duration and frequency, based on Equation 1. The results were then compared regionally to determine the appropriate factors to be applied to the site.

Regional Trends in Duration Factors

The duration factors for each station were plotted against return period (Figure 3-3). The Burwash A station was found to behave differently from the other four stations, as shown in Figure 3-3. The Pearson distribution was used to determine whether the duration factors for each duration were related to the elevation, latitude, or longitude of the meteorological station. Regional analysis suggests this is likely due to the stations high elevation and geographic position within a valley immediately on the leeward side of the Saint Elias Mountains, with a greater orographic influence than the other stations and the project area, which are located further east within the lower elevation interior of Yukon. Consequently, the longitude of the site, −137.2 degrees, was used to establish the duration factors for each return period at site. Table 3-3 and Figure 3-3 present the duration factors established for each regional station and for the site.

Table 3-3: Regional Analysis of Duration Factors

Station ID	Long.	Lat.	Elev. masl	Dist, km	Storm Duration								
					6 min	10 min	30 min	1 hr	2 hr	3 hr	6 hr	12 hr	24 hr
Mayo A	-135.87	63.62	503.80	131.10	0.20	0.26	0.29	0.34	0.37	0.45	0.61	0.78	1.00
Pelly Ranch	-137.37	62.82	454.20	23.45	0.23	0.32	0.40	0.48	0.58	0.66	0.75	0.85	1.00
Dawson A	-139.13	64.04	370.30	184.33	0.30	0.38	0.42	0.51	0.57	0.65	0.72	0.87	1.00
Carmacks	-136.30	62.10	524.90	75.35	0.24	0.34	0.40	0.47	0.52	0.60	0.72	0.89	1.00
Burwash A	-139.05	61.37	806.20	168.10	0.09	0.13	0.16	0.21	0.28	0.34	0.65	0.77	1.00
Site Estimated	-137.22	62.62	784.45	n/a	0.24	0.32	0.38	0.45	0.52	0.59	0.70	0.85	1.00

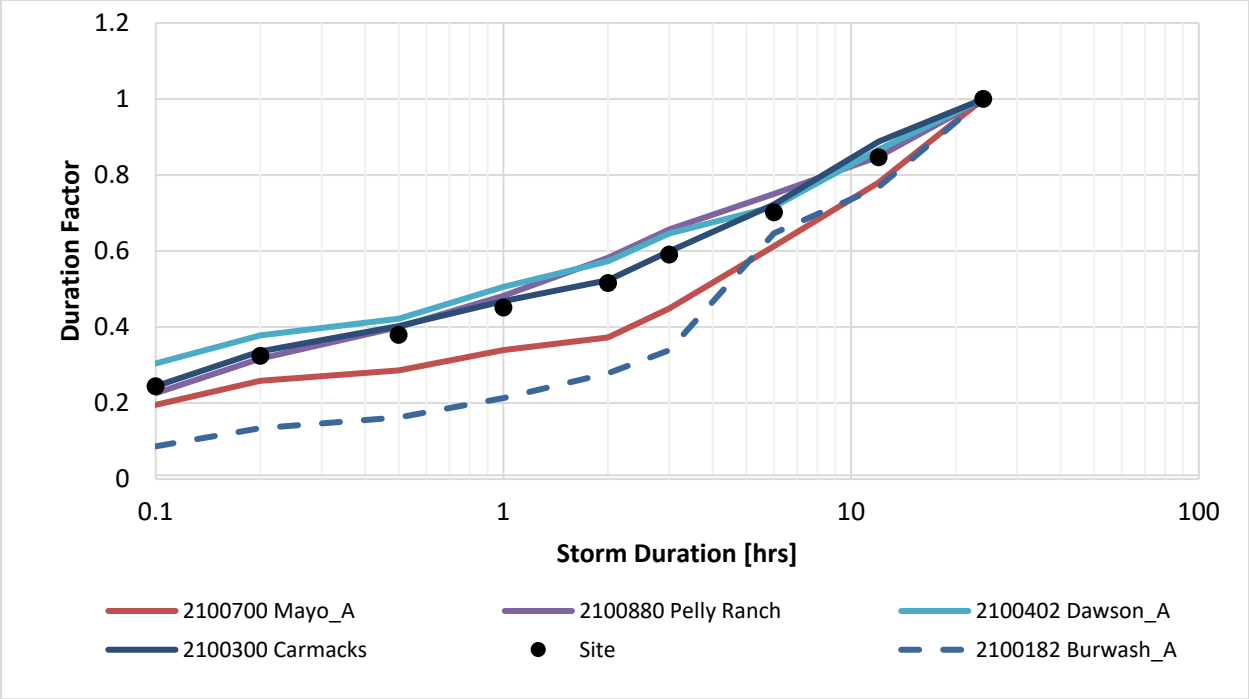


Figure 3-3: Regional and Site-Specific Duration Factors

Regional Trends in Frequency Factors

Similarly to the duration factors, the Pearson distribution was used to determine whether the frequency factors for each return period were related to the elevation, latitude, or longitude of the meteorological station. The best relationship was found to be with elevation; however, the difference in frequency factors across the region was minimal, and the site elevation was within the range of the stations analyzed. Therefore, the average from the regional frequency factors was applied to the site. Table 3-4 and Figure 3-4 present the duration factors established for each regional station and for the site.

Table 3-4: Regional Analysis of Frequency Factors

Station ID	Long. [degrees]	Lat. [degrees]	Elevation [masl]	Distance [km]	Return Period [years]					
					2	5	10	25	50	100
Mayo A	−135.87	63.62	503.80	131.10	0.70	0.88	1.00	1.15	1.26	1.37
Pelly Ranch	−137.37	62.82	454.20	23.45	0.58	0.83	1.00	1.19	1.34	1.49
Dawson A	−139.13	64.04	370.30	184.33	0.65	0.85	1.00	1.15	1.28	1.40
Carmacks	−136.30	62.10	524.90	75.35	0.64	0.85	1.00	1.17	1.30	1.43
Burwash A	−139.05	61.37	806.20	168.10	0.71	0.89	1.00	1.15	1.26	1.36
Site (Estimated)	−137.22	62.62	784.45	n/a	0.66	0.86	1.00	1.16	1.29	1.41

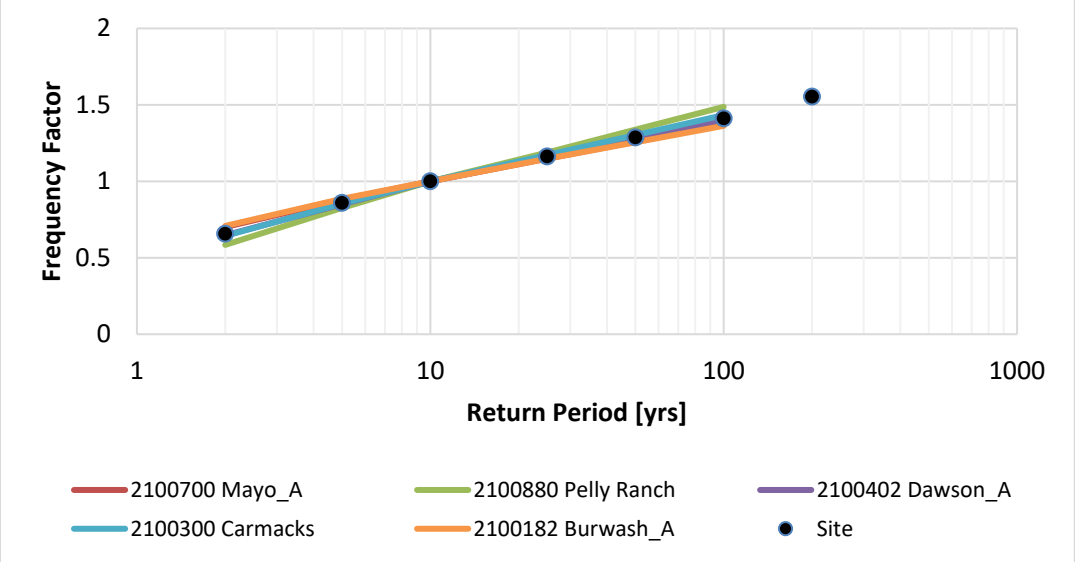


Figure 3-4: Regional and Site-Specific Frequency Factors

3.2.3 Site-Specific Intensity-Duration-Frequency Table

The results of site-specific frequency and duration factors and the 1:10 year, 24-hour duration rainfall of 34.2 mm were converted to an IDF table using Equation 1. In addition, the 1:200 year rainfall depth was extrapolated for each duration using a linear extrapolation of the duration factors and frequency factors. Results are presented in Table 3-5.

Table 3-5: Site-Specific IDF Table

Storm Duration		Return Period [years]						
Minutes	**Hours**	**2**	**5**	**10**	**25**	**50**	**100**	**200**
5	0.08	5.5	7.2	8.3	9.7	10.7	11.8	13.0
10	0.17	7.3	9.5	11.1	12.9	14.3	15.6	17.2
15	0.25	8.5	11.2	13.0	15.1	16.7	18.3	20.2
30	0.5	10.1	13.3	15.4	17.9	19.9	21.8	24.0
60	1	11.6	15.2	17.6	20.5	22.7	24.9	27.4
120	2	13.3	17.4	20.2	23.5	26.0	28.5	31.4
360	6	15.8	20.6	24.0	27.9	30.9	33.9	37.3
720	12	19.0	24.9	29.0	33.6	37.3	40.9	45.0
1440	24	23.5	30.0	34.2	39.3	42.9	46.6	47.6

3.2.4 Climate Change and IDF Results

The values for the respective return periods and durations were increased by 44% to reflect the forecasted increase resulting from the effects of future climate change (SRK 2016). Adjusted IDF values are presented in Table 3-6.

Table 3-6: Site-Specific IDF Table Adjusted for Total Precipitation Increase by 2100

Storm Duration		Return Periods [years]						
Minutes	**Hours**	**2**	**5**	**10**	**25**	**50**	**100**	**200**
5	0.08	7.9	10.3	12.0	14.0	15.5	17.0	18.7
10	0.17	10.5	13.7	16.0	18.5	20.5	22.5	24.8
15	0.25	12.3	16.1	18.7	21.7	24.0	26.4	29.0
30	0.5	14.6	19.1	22.2	25.8	28.6	31.4	34.5
60	1	16.7	21.8	25.4	29.5	32.7	35.8	39.5
120	2	19.1	25.0	29.1	33.8	37.4	41.1	45.2
360	6	22.7	29.7	34.6	40.2	44.5	48.8	53.7
720	12	27.4	35.8	41.7	48.5	53.7	58.9	64.8
1440	24	33.8	43.2	49.2	56.6	61.8	67.1	68.5

3.3 Snowmelt

Snowmelt estimation was performed in RStudio software using a daily snowmelt-energy model. This model incorporates daily meteorological parameters including daily maximum and minimum temperatures, wind speed, and total precipitation. Daily values for each parameter were obtained using the ERA-Interim reanalysis tool with daily information from 1979 to 2015 (ECMWF 2015).

Using the energy-based model (Walter 2005), the snow-water equivalent, potential snowmelt, and snowpack depth were calculated at each time step. The maximum daily snowmelt for each month of the simulated record from ERA-Interim was calculated, and the 50th percentile of maximum daily snowmelt was determined to be 23 mm/day (SRK 2016). This value can be applied as a baseflow on top of the design storm to estimate true peak flows for each catchment.

3.4 Probable Maximum Precipitation

The PMP for a 24-hour duration event was evaluated using the Hershfield method (World Meteorological Organization (WMO), 2009), for the same 20 regional meteorological stations used in the analysis for MAP. The Hershfield methodology requires the daily maximum 24 hour precipitation for at least 10 years of record. The analysis was performed in RStudio software, generating a PMP estimate for each of the 20 regional stations. A regression analysis was performed against longitude, which was found to have a better correlation than latitude or elevation. The results are presented in Figure 3-5, which presents a 24-hour PMP for the site of 159 mm.

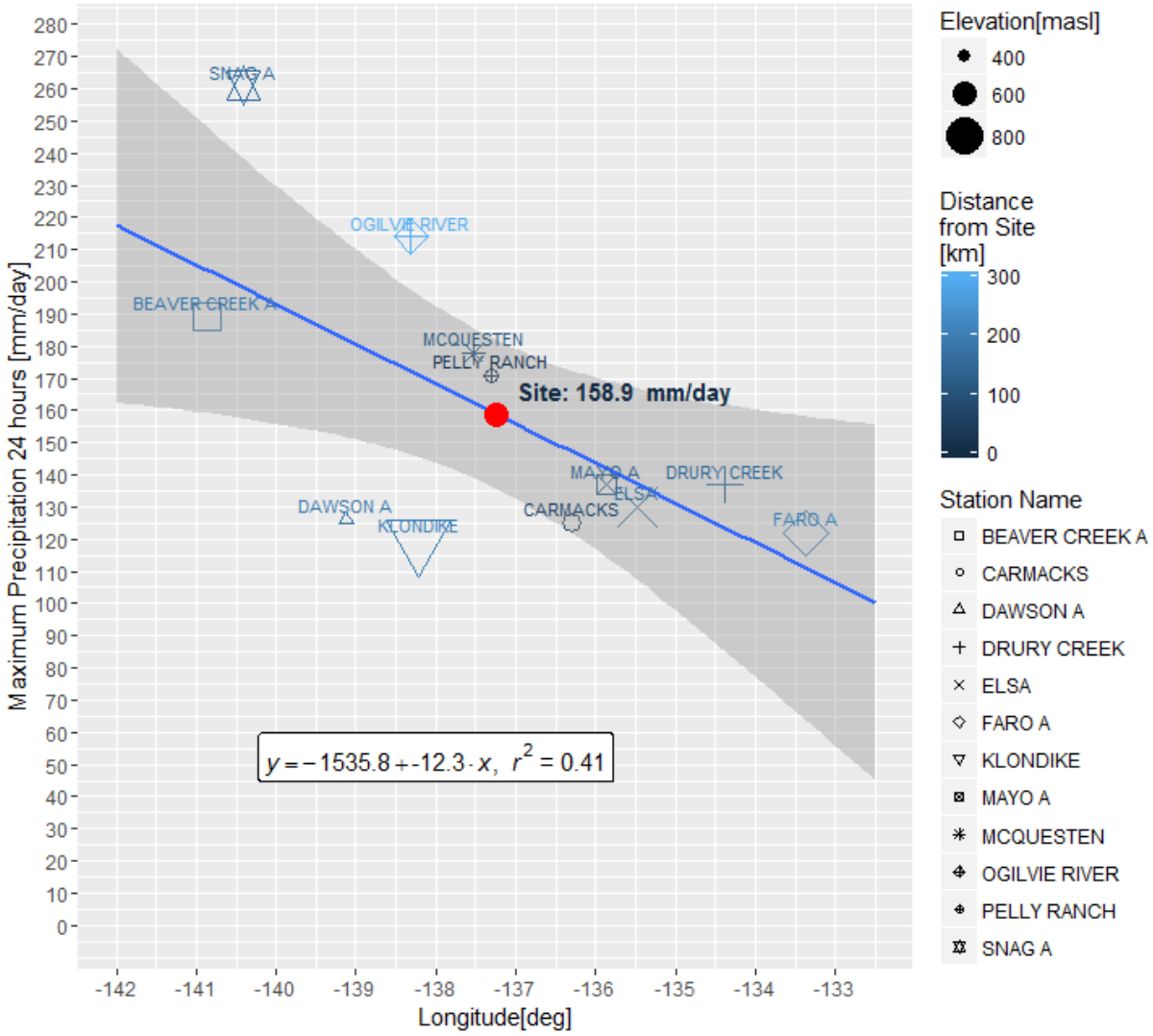


Figure 3-5: Regional Analysis for PMP

4 Peak Flows

A rainfall-runoff approach can be used to model peak flows at the site. Rainfall is applied to 24-hour distribution curve and is then converted to runoff based on catchment characteristics, including curve number, lag time, and total area, using the Soil Conservation Service (SCS) Method.

The following section summarizes the methodology to evaluate peak flows for the mine site.

4.1 Frequency-based Hypothetical Storm

The depth and temporal distribution of the storm hyetograph are defined using synthetic 24-hour rainfall distributions developed by National Resource Conservation Service (NRCS) (Chow et al. 1988), with a Type 1 rainfall distribution. This was selected due to the Pacific maritime climate with wet winters and dry summers observed in the Yukon, which closely mirrors the climate within Alaska for which a Type 1 distribution is typical (Chow et al. 1988). This distribution approximates an intense, short-duration storm event.

4.2 Initial loss

The SCS curve number model is used to estimate precipitation excess as a function of cumulative precipitation, soil cover, land use and antecedent moisture using the following equation:

$$P_e = \frac{(P - I_a)}{P - I_a + S}$$

Where,

P_e = accumulated precipitation excess at time *t*,

P = accumulated rainfall depth at time *t*,

I_a = the initial abstraction (initial loss), and

S= potential maximum retention, a measure of the ability of a watershed to abstract and retain storm precipitation.

An empirical relationship between I_a and S was developed by the SCS as:

$$I_a = 0.2S$$

The maximum retention, S, and watershed characteristics related through an intermediate parameter, the curve number was developed as:

$$S = \frac{25400 - 254\ CN}{CN}$$

Incremental excess for a time interval is computed as the difference between the accumulated excess at the end of and beginning of the period.

4.3 Hydrological Transformations

The SCS Unit Hydrograph method is used to transform precipitation excess into an outflow hydrograph for each sub-area. A HEC-HMS (2013) model can be used to evaluate peak flows. HEC-HMS uses a single input parameter, the lag time (T_{lag}), defined as the time between the centroid of precipitation mass to the peak for the resulting hydrograph. The lag time for each

catchment was calculated using the NRCS transformation from time of concentration. The NRCS transformation is T_{lag} = 0.6*Tc. Time of concentration and resulting lag times are relatively short (<15 minutes), resulting in a flashy response with peak discharge timing almost coincident with the timing of the precipitation peak. The time of concentration was estimated with the methods cited in (Li *et al*. 2008) where a minimum time of concentration of 10 minutes was applied.

4.4 Curve Number

The SCS curve number is utilized for the calculation of precipitation excess and the transformation of runoff within the watershed (NRCS 1986). The curve number for a watershed can be estimated as a function of land use, soil type, and antecedent watershed moisture, using tables published by the SCS and refined using calibration methods.

Once the closure period begins, the active mining areas will be decommissioned, and the land use is assumed to return to natural conditions. Based on this assumption, the curve number for the site can be validated using a regional catchment with known peak flows, a rainfall depth, and catchment characteristics including lag time and area.

A suitable regional catchment was selected using a regional analysis, comparing unit peak flows for eight nearby WSC hydrometric stations (EC 2015b). The stations are summarized in Table 4-1 (EC 2015b) and were compiled in RStudio software.

Table 4-1: Summary of Water Survey of Canada Stations used in Regional Analysis of Peak Flows

Station Name	Lat. [degrees]	Long [degrees]	Area [km^2]	Distance from Site [km]	Years with Info.
DRURY CREEK AT KM 469 ROBERT CAMPBELL HIGHWAY	62.2	-134.4	552	154	15
LITTLE SOUTH KLONDIKE RIVER BELOW ROSS CREEK	64.0	-137.6	860	154	13
GILTANA CREEK NEAR THE MOUTH	61.2	-137.0	190	159	27
SOUTH BIG SALMON RIVER BELOW LIVINGSTONE CREEK	61.4	-134.4	515	203	14
IBEX RIVER NEAR WHITEHORSE	60.7	-135.5	648	230	23
SIDNEY CREEK AT KILOMETRE 46 SOUTH CANOL ROAD	60.8	-133.1	372	301	12
WHEATON RIVER NEAR CARCROSS	60.1	-134.9	864	304	48
SOUTH MACMILLAN RIVER AT KILOMETRE 407 CANOL ROAD	62.9	-130.5	997	343	22
PINE CREEK NEAR ATLIN	59.6	-133.7	697	391	11

Peak and unit peak flows were calculated for return periods between the 1:2 year and 1:200 year events, where the unit peak flow is equal to the peak flow divided by the catchment area. Relationships were developed for unit peak flows of each return period and longitude, elevation and latitude. The best correlation was found to be with latitude, and the results of the regression analyses are presented in Figure 4-3.

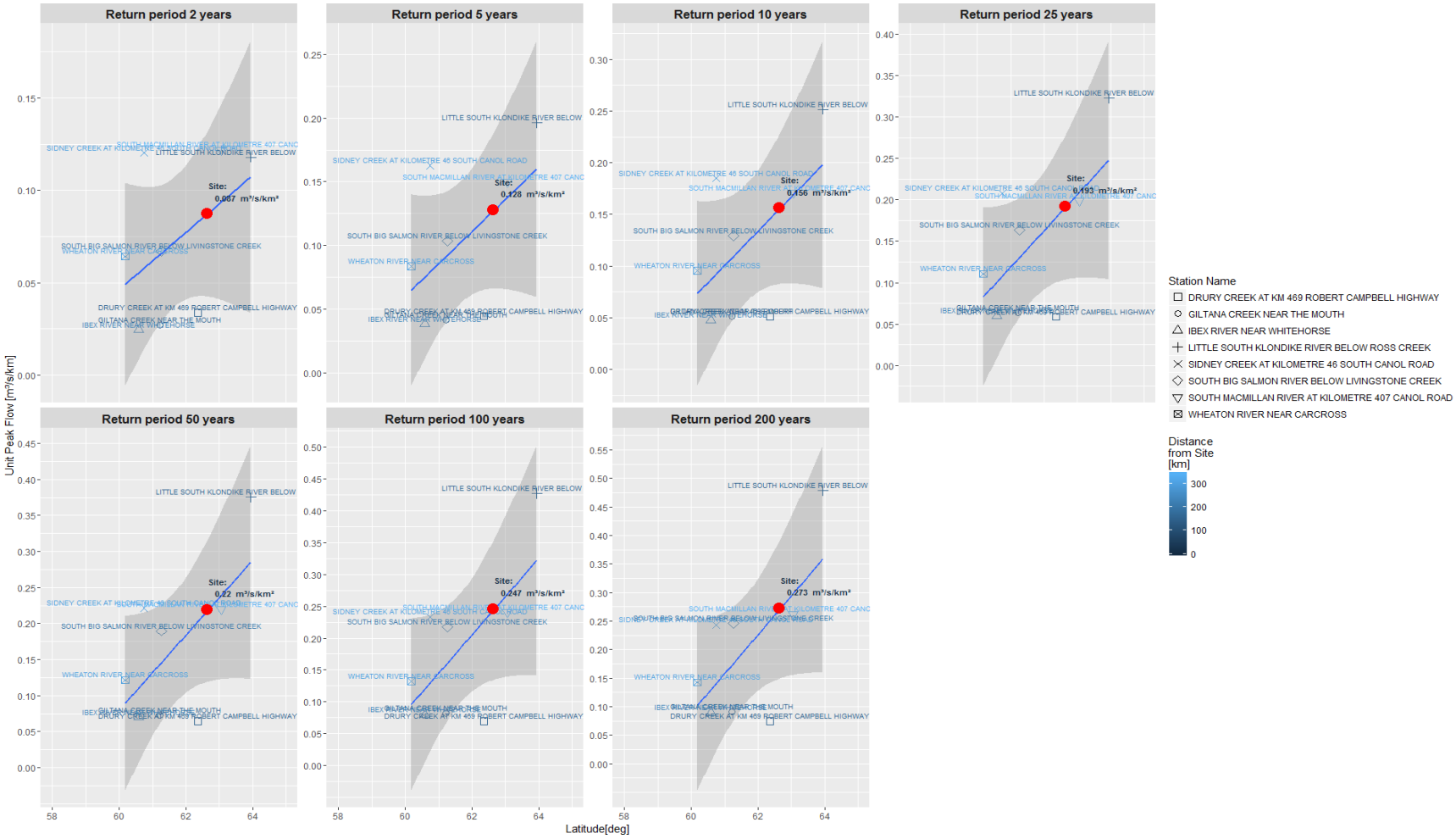


Figure 4-1: Regional Unit Peak Flows

For return periods greater than the 1:5 year event, the Little South Klondike River below Ross Creek presents the highest unit peak. For this reason, the Little South Klondike River below Ross Creek hydrometric station was selected for curve number calibration purposes to generate a conservatively high estimate.

The lag time for Little South Klondike River below Ross Creek was calculated to be 137 minutes, using the methodology described in Section 4.3 and was modelled in HEC-HMS (2013) along with the catchment area. Rainfall depths from the IDF results in Table 3-5 prior to climate change impacts were modelled for the 1:200 year event and the 1:10 year event, which form the design criteria for the closure conveyance structures. The curve number value was adjusted until the modelled peak flows matched the historically estimated peak flows. The results of the calibration are presented in Table 4-2, and include varying the curve number value from 70 to 84.

Table 4-2: Curve Number Calibration Results

Return Period [years]	Historic Rainfall Depth [mm]	Historic Peak Flow [m^3/s]	Curve Number					
			70	75	80	82	83	84
200	47.6	411.7	91.4	172.1	311.3	385.2	425.6	468.6
10	34.2	216.3	27.8	53	111.4	151	175.1	201.6

The 1:200 year return period was best matched using a curve number value of 83. The peak flow for the 1:10 year event using this curve number value is underestimated. However this return period is only used for the sizing of pilot channels within larger floodplain channels, ensuring that any overtopping of the pilot channel as a result of slightly underestimated flows will be captured. The curve number value of 83 was applied to all site catchments for closure conditions.

5 Summary

For the purpose of the post-closure water conveyance analysis, a hydrologic study was completed. The study used regional meteorological stations to generate expected precipitation estimates for the site that can be used for the design of the required conveyance infrastructure.

The following is a summary of the hydrology analysis for conveyance channels:

- Mean annual precipitation of 395 mm;
- The intensity-duration-frequency results are adjusted for potential impacts due to climate change, for a range of storm durations and frequencies (Table 3-6)
  - 1:200 year 24-hour rainfall depth of 68.5 mm,
  - 1:10 year 24-hour rainfall depth of 49.2 mm;
- Maximum expected 24-hour snowmelt depth of 23 mm including the impacts of climate change;

- 24-hour probable maximum precipitation (PMP) of 159 mm;
- Calibrated regional curve number for natural catchments is a value of 83.

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The opinions expressed in this report have been based on the information available to SRK at the time of preparation. SRK has exercised all due care in reviewing information supplied by others for use on this project. Whilst SRK has compared key supplied data with expected values, the accuracy of the results and conclusions from the review are entirely reliant on the accuracy and completeness of the supplied data. SRK does not accept responsibility for any errors or omissions in the supplied information, except to the extent that SRK was hired to verify the data.

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Appendix C – Climate Change Analysis

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Memo

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From:	Victor Munoz Saavedra Tia Shapka-Fels	**Project No:**	1CM002.042
Cc:	Dylan McGregor, SRK	**Date:**	August 3, 2016
Subject:	Climate Change Analysis, Minto Mine, Yukon		

1 Introduction

SRK Consulting (Canada) Inc. has prepared an updated closure water conveyance plan design report for Minto Mine (Minto) located in the Yukon, Canada, and operated by Minto Exploration Ltd.

There is agreement within the engineering community that climate change is occurring and that climate change needs to be integrated into engineering designs. Engineers Canada (2013) encourages its members to keep themselves informed about the changing climate and consider potential impacts on their professional activities. Design integration of climate change analysis results looks to the long term performance of mine closure designs such as conveyance channels and waste rock cover systems. Mine closure infrastructure designed using average daily temperatures and precipitation values during mines operation may not be adequate for infrastructures that support the mine site in the future when the climate has changed.

This memo presents climate change projections of key climatic and hydrological parameters which may be experienced at the mine site in the future. Parameters include, but are not limited to, projections for changes in temperature, rain and snowfall precipitation. This memo further discusses how the climate change projections for the Minto site are then integrated into the closure conveyance channel design.

2 Method

Climate change analysis includes collating and evaluating baseline climate data, querying available climate change prediction models, and making forecasting climate trends. This analysis produced a graphical summary of results (Section 5). The summaries presented in Section 5 are a probabilistic expected change in the climatic with respect to baseline conditions at the Minto mine site.

The analysis required three steps as outlined in Section 2.1 to 2.3.

2.1 Step 1: Data Retrieval

The first step of the analysis was to identify the availability of the climate change models for each required climate parameter. Environment Canada provides access to monthly climate change modeling predictions for any location in the World from the following climate change assessment reports:

- Canadian Regional Climate Model (CRCM), also called First Assessment Report (AR1) (EC 2015)
- Second Assessment Report (AR2) (IPCC 1995)
- Third Assessment Report (AR3) (IPCC 2001)
- Fourth Assessment Report (AR4) (IPCC 2007)
- Fifth Assessment Report (AR5) (IPCC 2014)

Information for the first four assessment reports is available from 1960, with predictions out to 2100 for the meteorological variables listed in Table 2.1; however, there are significant gaps depending on the assessment report, scenario evaluated, and variable assessed. Information for the fifth assessment report is available for 1900 to 2100, but only includes mean temperature and total precipitation.

As part of the procedure, data from the five assessment reports are downloaded from Environment Canada for the site being evaluated based on latitude and longitude. This process is automated through a purpose built script developed using R programming language (CRAN 2015).

The meteorological parameters covered in the assessment reports are used, though application of empirical models, to calculate climate change design parameters for other parameters such as intensity-duration–frequency (IDF) curves (i.e. extreme storm events), snowpack thickness, evaporation, etc.

Table 2.1: Metrological Parameters Listed in AR1 through AR4

Parameter	Base Parameter	Details
Air Temperature	Temperature	Mean (T_{mean}), Mean Max (T_{max}), Mean Minimum (T_{min}), and Extreme Range at 2 m above ground surface
Heat Wave Duration Index	Temperature	Maximum period >5 consecutive days with T_{max} >5°C above the baseline T_{max} normal
Frost Days	Temperature	Total number of frost days (days with absolute minimum temperature <0°C)
Growing Season Length (GSL)	Temperature	Growing season length is defined as the time period each year between when T_{mean} >5°C for more than 5 days and when T_{mean} <5°C for more than 5 days
Air Temp. Extreme Range	Temperature	Intra-annual extreme temperature range is the difference between the highest temperature and lowest temperature of the same calendar year

Parameter	Base Parameter	Details
Total Precipitation	Precipitation	Total precipitation including rainfall and snowfall as snow water equivalent
Days with rain >10 mm	Precipitation	Days with rain greater than or equal to 10 mm/day
Simple Daily Intensity Index	Precipitation	Annual total precipitation divided by the number of precipitation days greater than or equal to 1 mm/day
Dry Days	Precipitation	Max amount of consecutives dry days (precipitation <1 mm)
Fraction of total annual precipitation >95th percentile	Precipitation	Annual precipitation >95th percentile
Wind Speed	Wind Speed	Mean, Meridional, and Zonal wind speed at 10 m above ground surface
Solar Radiation	Radiation	Shortwave surface down-welling
Humidity	Humidity	Relative and specific humidity at 2 m above ground surface
Sea Level Pressure	Air Pressure	Mean sea level pressure

2.2 Step 2: Baseline Analysis (Outcome 1)

Climate change models presented in the assessment reports assume application of radiative forces (energy flux) through different anthropogenic sources that results in discharge of varying concentrations of atmospheric greenhouse gases. These radiative forces are not constant through time, as they are based on global anthropogenic behavior such as environmental policies, population growth, economic growth, energy sources, land use, hydrocarbon usage, etc. Each individual climate change model presented in the assessment reports represent these non-linear radiative forces differently and thus present a different climate change scenario, underscored by its own model assumptions and boundary conditions.

None of these individual models are inherently superior or inferior to others, and likewise, the newer generation of assessment reports are not necessarily more reliable than older versions, but rather represent more detailed consideration of global anthropogenic forces. Therefore, the user is forced to make a judgement call on which individual model or generation of models are the most suitable to use in design, which invariably leads to bias. This method of analysis is specifically aimed at eliminating this bias by analysing all the available climate change models with equal weight and presenting a single climate change design parameter based on a rational statistical evaluation of the overall cumulative results.

Using each and every climate change model available in the five assessment reports, the projected change with respect to a set baseline condition can be calculated using Equation 2.1.

Equation 2.1

$$\textbf{\textit{Change with respect to set baseline condition}}\ [\%] = \frac{\textbf{\textit{Projection}} - \textbf{\textit{Baseline}}}{\textbf{\textit{Baseline}}}$$

In Equation 2.1, the *Baseline* and *Projection* periods have a minimum timeframe of 30 years, which is generally recognized as being the minimum acceptable period length of record that is statistically significant (WMO 2007). These periods can be defined as follows:

- *Baseline* is the time frame associated with the past (i.e. estimation of the climate data based on the climate change models). For the purpose of this analysis, this period is fixed from 1975 to 2005, which coincides with the baseline adopted for AR5.
- *Projection* is the future time frame where climate change models are applied, reflecting climate change predictions. Three *Projection* periods are applied from 2011 to 2040, 2041 to 2070, and 2071 to 2100. These 30-year periods are based on convention as stipulated in WMO (2007).

Climate change model predictions are currently presented up to the year 2100 (IPCC 2014), which is deemed that maximum reasonable timeframe to extend predictions to. As a result, this is the maximum timeframe for which this analysis method applies.

Equation 2.1 is automatically processed using a script developed in R. An example result is presented in Figure 2.1, which represents the outcome at a hypothetical site for the meteorological parameter mean air temperature. For this example, the *Projection* period is representative of 2041 to 2070.

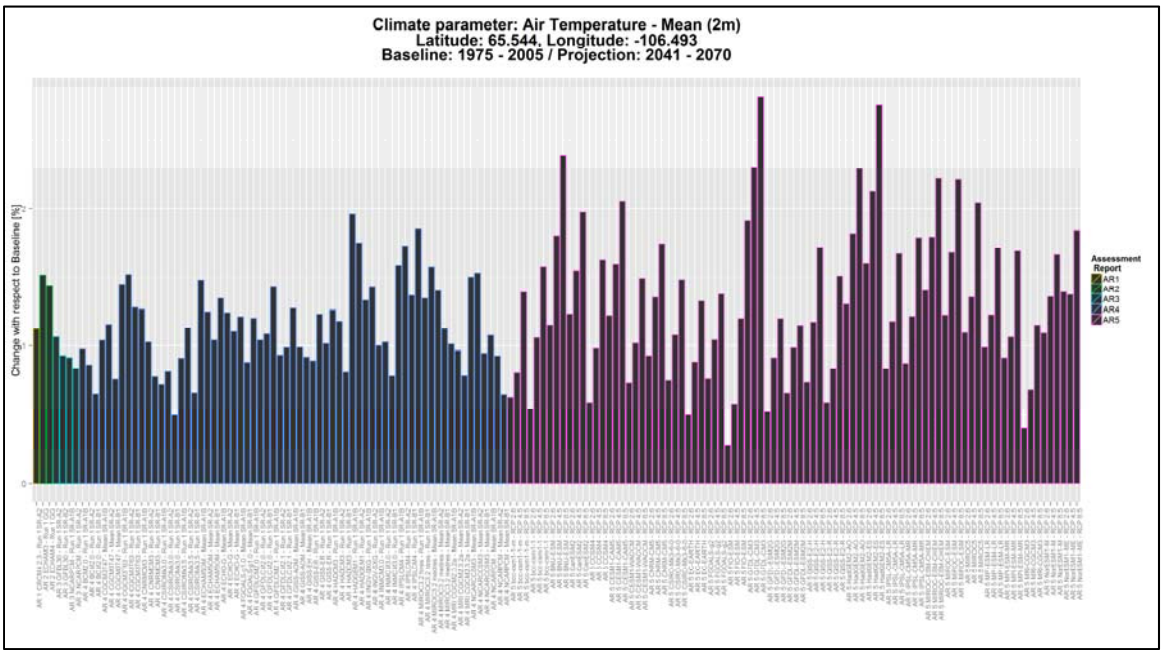


Figure 2.1: Example outcome for hypothetical site showing results of change to baseline conditions for individual climate change models associated with each of the five assessment reports for a projection period from 2041 to 2070

The analysis presented in Figure 2.1 can be presented as a box-whisker plot as illustrated in Figure 2.2. The change with respect to baseline conditions for each individual Assessment Report is summarized, together with an overall summary representing all the assessment reports combined. The centerline of the box represents the median value, while the upper and lower box borders represent the first and third quartile values, respectively. The limits whiskers will extend to the maximum and minimum of 1.5 times the total range of the box (i.e. the third quartile minus first quartile). If values exist outside the whisker limits (i.e. outliers), then they are presented as dots.

The results in Figure 2.2 can also be presented as a cumulative probabilistic curve as illustrated in Figure 2.3. To weigh all climate change models equally, only the cumulative probabilistic curve associated with combined data from all assessment reports was considered in this analysis. The overall spread of results as presented by the collection of box-whisker plots in Figure 2.2 illustrates why this is deemed appropriate.

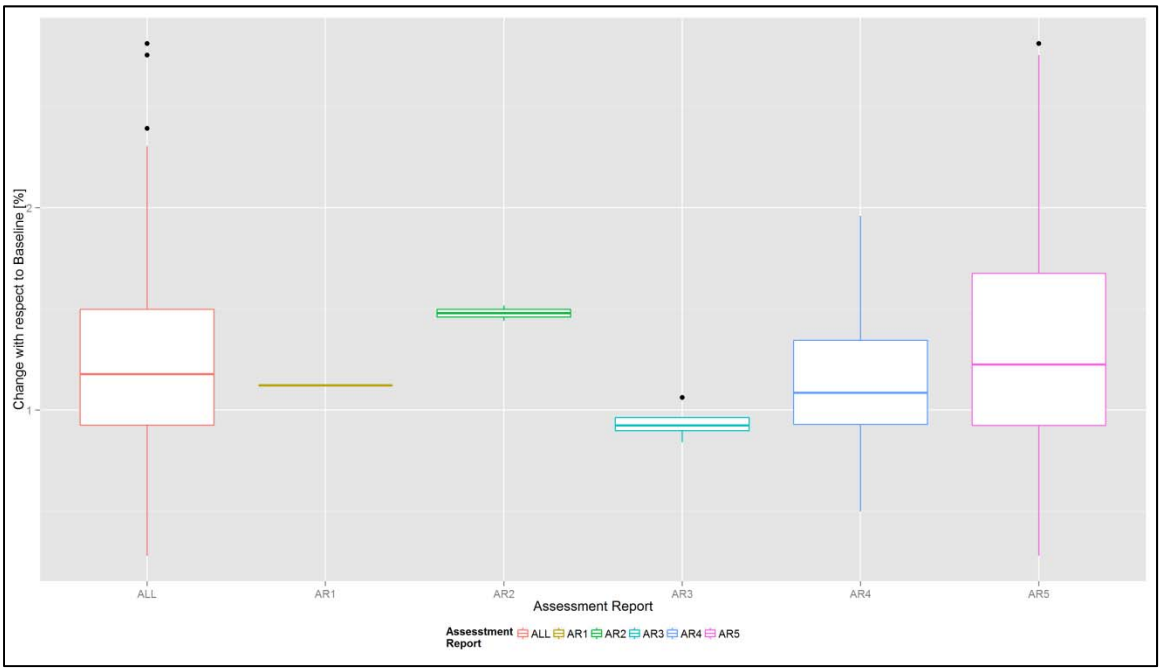


Figure 2.2: Combined box-whisker plot of change to baseline conditions for the individual climate change models presented in the Figure 2.1 example

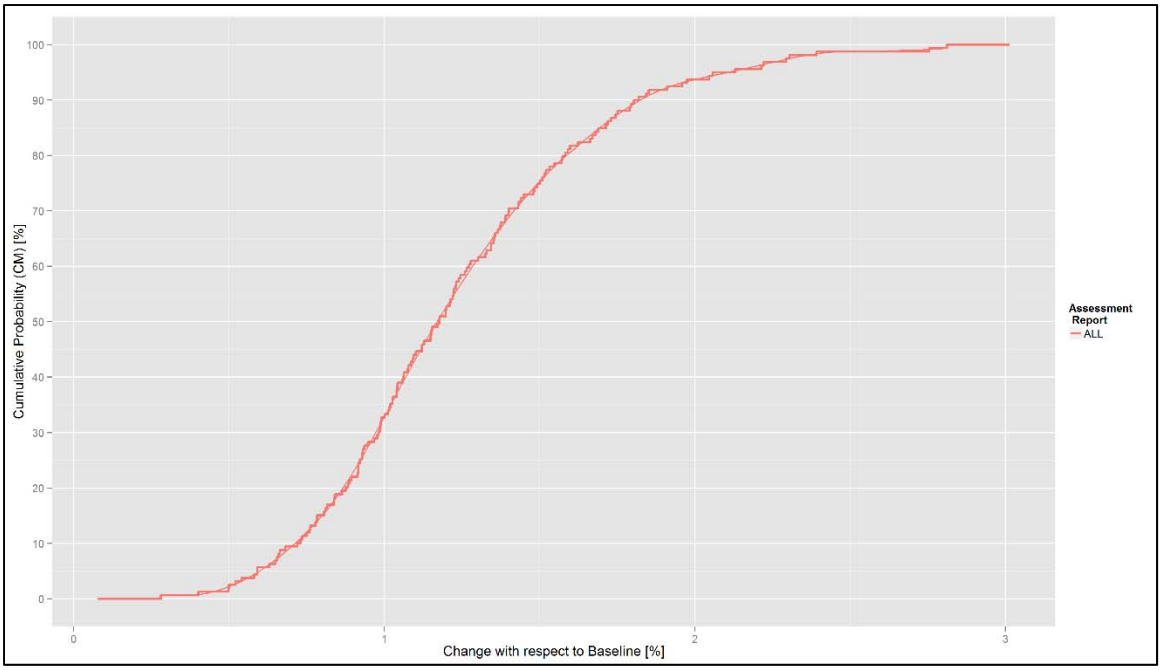


Figure 2.3: Cumulative probability curve of climate change prediction models from all assessment reports expressed as a percentage of change against baseline conditions

2.3 Step 3: Trend Analysis (Outcome 2)

As discussed in Section 2.2, climate change models consider global anthropogenic forces that are not constant in time. Irrespective of such forces, the science of hydrology estimates and predicts future meteorological conditions through consideration of historic trends. Therefore, to facilitate selection of climate change parameters for design, a trend analysis approach is used as a reality check.

To best represent the historical trend, a global climatic model reanalysis approach is used, rather than relying on regional meteorological stations for historic data, since the availability and timespan of records vary greatly. Reanalysis combines satellite information, land records, and numerical models that simulate the earth's climatic conditions. Typically, reanalysis extends for several decades and covers the entire planet. State-of-the-art publically available reanalysis data from ERA-Interim produced by the European Center for Medium-Range Weather Forecast (ECMWF 2015) was used in this analysis. ERA-Interim includes twice daily data from 1979 to 2015 for the entire world, based on a 0.75 degree latitude by 0.75 degree longitude grid.

The ERA-Interim data trend analysis entailed two parts: first, identification of the trend, and second, estimation of the statistical significance of the trend. Five different trend analysis methodologies were considered:

- Ordinary least square (Maidment 1993)
- Quantile regression (Koenker and Bassett 1978)

- Mann-Kendall and Theil Sen (Mann 1945 and Sen 1968)
- Zhang (Zhang 2000)
- Yue and Pilon (Yue et al 2002)

Figure 2.4 presents the outcome of the reanalysis for the hypothetical example site presented in Figure 2.1 through Figure 2.3. Each regression method has a statistical significance value over 95%; however, this may not always be the case and the meaning of that is described further in next section.

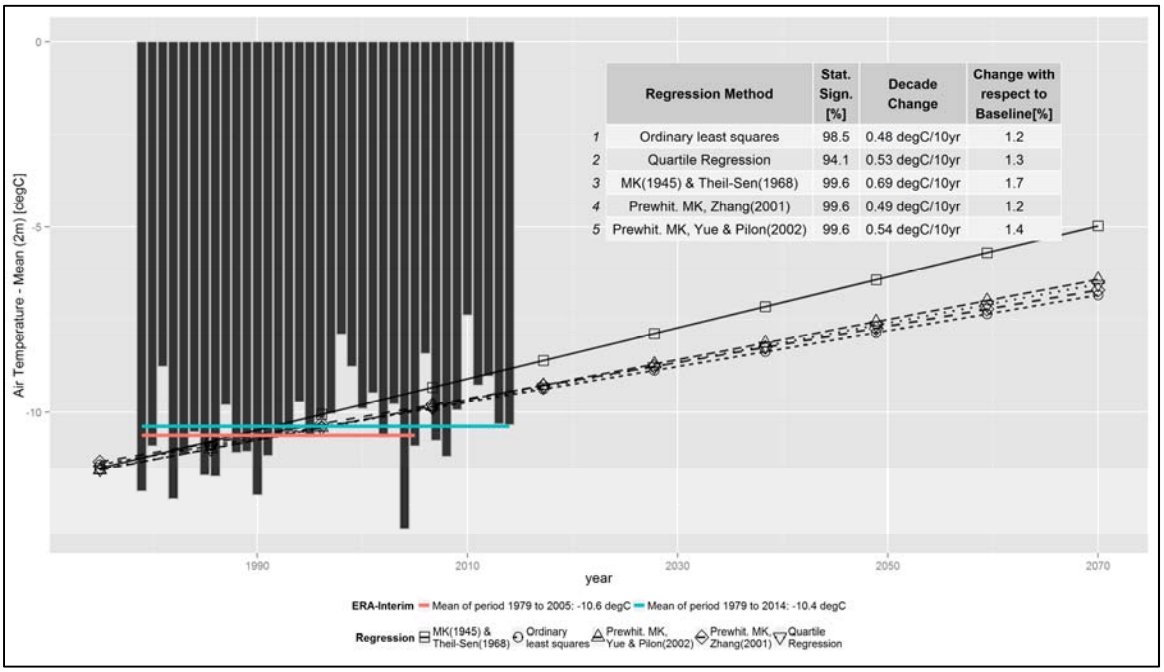


Figure 2.4: Trend analysis outcome based on reanalysis evaluation for hypothetical site presented in Figure 2.1

2.4 Recommended Climate Change Design Parameter

Following completion of the previously described procedure, a design recommendation for each identified climate change parameter is presented. The previous procedure produces two outcomes as follows:

- Outcome 1 is a cumulative probability curve representing change relative to baseline conditions based on the Step 2 analysis; this is also represented by Figure 2.3.
- Outcome 2 is the five separate regression analysis results from the Step 3 trend analysis, again representing change relative to baseline conditions represented by Figure 2.4. The statistical significance of each regression analysis is calculated during this step and must yield a value greater than 95% to be deemed statistically significant.

If the trend analysis shows no regression of statistical significance (i.e. greater than 95%), the climate change design parameter is the 50% cumulative probability value based on the Step 2 baseline analysis. If on the other hand, one or more of the trend analyses suggest trends of statistical significance, the climate change parameter is the maximum of either the 50% cumulative probability value based on the Step 2 baseline analysis, or the mean of all the trend analysis values with statistical significance. Equation 2.2 provides an expression for this criteria.

Equation 2.2

$$Recommended\ Climate\ Change\ Design\ Parameter = \text{Maximum}(50\%\ Cumulative\ Probability,\ Mean\ \{Regression_{Stat.\ Sign \geq 95\%}\})$$

3 Climate Change Projections

3.1 Long Term Temperature Trends

The mean annual air temperature at Minto is expected to increase by about 3.3°C over the next century (Table 3.1), with mean annual air temperatures of −0.8°C, 0.2°C, and 1.1°C in the 2020s, 2050s and 2080s, respectively. These predictions compare well with those presented by other studies. The International Panel on Climate Change (IPCC) projects a 5°C to 6°C increase in the temperature in Northern Canada for 2100 relative to the 1980 to 1999 temperatures (IPCC 2007).

Table 3.1 though Table 3.8 present additional temperature related climate forecasts for the Minto area. These tables predict longer periods with high temperatures and consequently shorter periods with lower temperatures and less inter-annual temperature variation.

Table 3.1: Mean Annual Air Temperature at Minto

Timeline	Context	Value (°C)	Change Over Baseline (°C)	Change [%]
1979-2005	Baseline	-2.12	-	-
1979-2014	Current Reality	-2.29	-0.17	-0.06%
2011-2040	2020s	-0.82	1.30	0.48%
2041-2070	2050s	0.24	2.36	0.87%
2071-2100	2080s	1.13	3.25	1.20%

Table 3.2: Mean Maximum Air Temperature at Minto

Timeline	Context	Value (°C)	Change Over Baseline (°C)	Change [%]
1979-2005	Baseline	1.39	-	-
1979-2014	Current Reality	1.26	-0.13	-0.05%
2011-2040	2020s	1.40	0.01	0.36%
2041-2070	2050s	1.40	0.01	0.69%
2071-2100	2080s	1.41	0.02	1.10%

Table 3.3: Mean Minimum Air Temperature at Minto

Timeline	Context	Value (°C)	Change Over Baseline (°C)	Change [%]
1979-2005	Baseline	-5.46	-	-
1979-2014	Current Reality	-5.68	-0.22	-0.08%
2011-2040	2020s	-5.49	-0.03	0.48%
2041-2070	2050s	-5.51	-0.05	0.93%
2071-2100	2080s	-5.54	-0.08	1.40%

Table 3.4: Intra-Annual Temperature Range at Minto(1)

Timeline	Context	Value (°C)	Change Over Baseline (°C)	Change [%]
1979-2005	Baseline	61.3	-	-
1979-2014	Current Reality	62.2	0.9	0.27%
2011-2040	2020s	59.8	-1.5	-0.45%
2041-2070	2050s	58.6	-2.7	-0.81%
2071-2100	2080s	57.6	-3.7	-1.10%

Note(s):

(1) The difference between the highest temperature of the year and the lowest temperature of the year.

Table 3.5: Daily Minimum Air Temperature, 90th Percentile at Minto

Timeline	Context	Value (%)	Change Over Baseline (%)	Change [%]
1979-2005	Baseline	10.0	-	-
1979-2014	Current Reality	10.2	0.2	2.0%
2011-2040	2020s	14.6	4.6	46.0%
2041-2070	2050s	21.0	11.0	110.0%
2071-2100	2080s	31.0	21.0	210.0%

Table 3.6: Growing Season Length(1) at Minto

Timeline	Context	Value (days)	Change Over Baseline (days)	Change [%]
1979-2005	Baseline	93	-	-
1979-2014	Current Reality	95.9	2.9	3%
2011-2040	2020s	115.3	22.3	24%
2041-2070	2050s	136.7	43.7	47%
2071-2100	2080s	147.9	54.9	59%

Note(s):

(1) Growing season length is the number of days between the first five consecutive days of the year with a mean air temperature above 5°C, and the last five such days of the year.

Table 3.7: Heat Wave Duration Index(1) at Minto

Timeline	Context	Value (days)	Change Over Baseline (days)	Change [%]
1979-2005	Baseline	108	-	-
1979-2014	Current Reality	110	2.0	2%
2011-2040	2020s	143	34.6	32%
2041-2070	2050s	185	76.7	71%
2071-2100	2080s	270	162.0	150%

Note(s):

(1) The maximum period for the year of at least five consecutive days with the maximum air temperature at least 5°C warmer than the daily climatology baseline.

Table 3.8: Frost Days(1) at Minto

Timeline	Context	Value (days)	Change Over Baseline (days)	Change [%]
1979-2005	Baseline	225	-	-
1979-2014	Current Reality	225	0.0	0.0%
2011-2040	2020s	216	-9.5	-4.2%
2041-2070	2050s	207	-18.5	-8.2%
2071-2100	2080s	196	-29.3	-13.0%

Note(s):

(1) Total number of days with an absolute minimum temperature less than 0°C.

3.2 Long Term Precipitation Trends

The total precipitation at Minto is forecasted to increase 67.9 mm (15%) during the mine life (2011 to 2040) and 199.3 mm (44%) by 2100 (Table 3.9).

Table 3.9: Total Precipitation at Minto

Timeline	Context	Value (mm)	Change Over Baseline (mm)	Change [%]
1979-2005	Baseline	453	-	-
1979-2014	Current Reality	469	16.0	4%
2011-2040	2020s	521	67.9	15%
2041-2070	2050s	584	131.4	29%
2071-2100	2080s	652	199.3	44%

3.3 Frequency of Drought

Climate change model suggests that consecutive dry days will decrease by up to 13% by 2100 (Table 3.10).The clear mean annual temperature increase trend (Table 3.1) along with the increasing precipitation trend (Table 3.9) and decreasing consecutive dry day trend (Table 3.10) suggest that the frequency of drought should be decreasing.

Table 3.10: Consecutive Dry Days(1) at Minto

Timeline	Context	Value (days)	Change Over Baseline (days)	Change [%]
1979-2005	Baseline	17.1	-	-
1979-2014	Current Reality	17.8	0.7	4.1%
2011-2040	2020s	16.0	-1.1	-6.4%
2041-2070	2050s	15.2	-1.9	-11.0%
2071-2100	2080s	14.9	-2.2	-13.0%

Note(s):

(1) Maximum number of consecutive dry day is a year. Days with less than 1 mm of precipitation are considered dry days.

3.4 Frequency of Rainfall

The simple daily precipitation intensity index is the mean annual precipitation divided by the total number of wet days. Therefore, the total number of wet days can be estimated from the simple daily precipitation intensity index values and the total precipitation.

The simple daily precipitation intensity index at Minto is forecasted to increase by 6.6% during the mine life and by 25% in the next century (Table 3.11).

Table 3.12 presents the estimated number of wet days. The total estimated number of wet days is forecasted to increase by a similar amount with a 7.9% increase during the mine life and a 15.2% increase during the next century.

Table 3.11: Simple Daily Precipitation Intensity Index(1) at Minto

Timeline	Context	Value (mm/year)	Change Over Baseline (mm/year)	Change [%]
1979-2005	Baseline	3.9	-	-
1979-2014	Current Reality	3.95	0.05	1.28%
2011-2040	2020s	4.16	0.26	6.60%
2041-2070	2050s	4.41	0.51	13.00%
2071-2100	2080s	4.88	0.98	25.00%

Note(s):

(1) The total annual precipitation divided by the total number of wet days in a year

Table 3.12: Estimated Total Number of Wet Days(1) at Minto

Timeline	Context	Value (days)	Change Over Baseline (days)	Change [%]
1979-2005	Baseline	116.2	-	-
1979-2014	Current Reality	118.7	2.6	2.2%
2011-2040	2020s	125.3	9.2	7.9%
2041-2070	2050s	132.6	16.4	14.2%
2071-2100	2080s	133.8	17.7	15.2%

Note(s):

(1) The total number of wet days is not directly developed from the R analysis, but rather calculated based on the values in Table 3.9 and Table 3.11.

3.5 Long Duration Rainfall

Long duration rainfall, fraction of annual precipitation greater than 95th percentile, is predicted to increase by 96% over the next century at Minto (Table 3.13). The number of days with more than 10 mm of rainfall is forecasted to increase by 140% (Table 3.14).

The change in days with more than 10 mm of rain is forecasted to be 58% respectively during mine life (2011 to 2040). Similarly the maximum total precipitation for a five day period will increase by 20% over the next century and 7.8% over the mine life (2011 to 2040) as shown in

Table 3.15. Therefore, the volume and frequency of high precipitation events will tend to increase over the century.

Table 3.13: Fraction of Annual Precipitation >95th Percentile(1) at Minto

Timeline	Context	Value (%)	Change Over Baseline (%)	Change [%]
1979-2005	Baseline	5.01	-	-
1979-2014	Current Reality	5.29	0.28	5.59%
2011-2040	2020s	7.01	2.00	40.00%
2041-2070	2050s	7.52	2.51	50.00%
2071-2100	2080s	9.82	4.81	96.00%

Note(s):

(1) Percentage of days with precipitation greater than the 95th percentile calculated for wet days from the baseline period.

Table 3.14: Days with Precipitation Greater Than or Equal to 10 mm at Minto

Timeline	Context	Value (days)	Change Over Baseline (days)	Change [%]
1979-2005	Baseline	3.88	-	-
1979-2014	Current Reality	4.3	0.38	9.79%
2011-2040	2020s	6.1	2.25	58.00%
2041-2070	2050s	7.7	3.84	99.00%
2071-2100	2080s	9.3	5.43	140.00%

Table 3.15: Greatest Five Days Total Rainfall at Minto

Timeline	Context	Value (mm)	Change Over Baseline (mm)	Change [%]
1979-2005	Baseline	38.9	-	-
1979-2014	Current Reality	40.2	1.3	3.3%
2011-2040	2020s	41.9	3.0	7.8%
2041-2070	2050s	43.6	4.7	12.0%
2071-2100	2080s	46.7	7.8	20.0%

3.6 Short Duration Rainfall

Based on an evaluation of the predicted values for greatest five-day total precipitation, consecutive dry days, simple daily intensity index, days with rain greater than 10 mm, and fraction of total annual precipitation greater than the 95th percentile, the IDF for Minto was adjusted, where the total precipitation increase is forecasted to increase by 44% in 2100. Using the current reality, the values for the respective return periods and durations were increased by 44% to reflect the forecasted increase

Table 3.16: IDF Adjusted for Total Precipitation Increase by 2100

Duration		Return Periods [years]						
Minutes	Hours	2	5	10	25	50	100	200
5	0.08	7.9	10.3	12.0	14.0	15.5	17.0	18.7
10	0.17	10.5	13.7	16.0	18.5	20.5	22.5	24.8
15	0.25	12.3	16.1	18.7	21.7	24.0	26.4	29.0
30	0.5	14.6	19.1	22.2	25.8	28.6	31.4	34.5
60	1	16.7	21.8	25.4	29.5	32.7	35.8	39.5
120	2	19.1	25.0	29.1	33.8	37.4	41.1	45.2
360	6	22.7	29.7	34.6	40.2	44.5	48.8	53.7
720	12	27.4	35.8	41.7	48.5	53.7	58.9	64.8
1440	24	33.8	43.2	49.2	56.6	61.8	67.1	68.5

3.7 Snow Accumulation

Estimations of snow accumulation were developed by coupling the climate change model predictions for precipitation, temperature, and wind speed with the energy snowmelt model by Walter et al. (2005).

To estimate snowmelt, a subroutine from the EcoHydRolody library in R, called SnowMelt was used to analyze snow melt from an energy-based model. This program function forecasts the future daily snowmelt using energy--based model posed by Walter et al. (2005). This hydrological

model is based on typical meteorological parameters (e.g. daily maximum, minimum temperature, wind speed, and total precipitation).

These inputs were obtained on a sub-daily and daily scale from the reanalysis source ERA-Interim from 1979 to 2015. As a terrain source, the model required latitude and longitude of the site, topographical slope, and aspect of the terrain. These last two parameters were obtained from the topographical information GTopo30. This source provides a worldwide topographical information with a spacing of 30 arc-sec which is approximately equal to 1 km.

The result of the model is a daily snowmelt model including parameters such as snowfall water equivalent, snowmelt, snowpack depth, and snow water equivalent of snowpack.

The monthly average snow depth at Minto is expected to increase from over the next century (Table 3.17). In the time periods 1979 to 2005 and 1979 to 2014, the snowpack depth will tend to increase with the time to a maximum in the 2100s. These snowpack values should be taken as a reference as snowmelt model should be calibrated with site information.

Table 3.17: Monthly Average Snowpack at Minto

Timeline	Snow Depth (mm)											
	Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec
1979-2005	230	260	240	220	30	0	0	10	30	100	160	180
1979-2014	240	260	260	230	40	0	0	10	30	110	170	210
2011-2040	270	280	280	230	30	0	0	0	20	100	180	230
2041-2070	290	300	280	210	20	0	0	0	10	90	190	240
2071-2100	330	330	300	200	10	0	0	0	10	90	210	280

3.8 Snowmelt

The daily maximum snowmelt in an average year at Minto can be seen in Table 3.18, the month in which these maximum values are obtained from can be seen in Table 3.19. These predictions forecast freshet in April, and slightly increasing maximum daily snowmelt over the next century. The predicted maximum daily snowmelt in 2071 to 2100 is similar that of the snowmelt 1979 to 2005 (Table 3.18).

Table 3.18: Maximum Daily Snowmelt in Average Year at Minto

Timeline	Daily Maximum Snowmelt (mm/day)
1979-2005	20
1979-2014	20
2011-2040	22
2041-2070	23
2071-2100	22

Table 3.19: Daily Maximum Snowmelt per Month at Minto

Timeline	Snow Melt (mm/day)											
	Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec
1979-2005	4	4	6	20	0	0	0	0	1	6	4	4
1979-2014	3	4	6	20	0	0	0	0	0	5	3	4
2011-2040	4	6	9	22	0	0	0	0	0	5	4	4
2041-2070	5	7	13	23	0	0	0	0	0	3	4	5
2071-2100	5	9	16	22	0	0	0	0	0	3	4	6

3.9 Wind Speed

The mean annual wind speed at Minto is predicted to increase by 2.4% over the next century (Table 3.20).

Table 3.20: Mean Wind Speed at Minto

Timeline	Context	Value (m/s)	Change Over Baseline (m/s)	Change [%]
1979-2005	Baseline	2.56	-	-
1979-2014	Current Reality	2.55	-0.01	-0.39%
2011-2040	2020s	2.56	0.00	0.01%
2041-2070	2050s	2.58	0.02	0.87%
2071-2100	2080s	2.62	0.06	2.40%

4 Benchmarking

The climate change projections were benchmarked against the following reports with climate projections:

- **Yukon Climate Change Indicators and Key Findings 2015** (NCEC 2015). The report is designed to provide an overview of the trends in temperature, precipitation, fire severity, and sea ice extent/volume for decision makers in the province. The temperature and precipitation trends can be used as a comparative for the climate change value evaluated in this report.

- **Coffee Gold Project, Climate Change Projection for Coffee Creek Region, Yukon** (Lorax 2015). The report is designed to address climate projections and their potential impacts on the project. They address temperature, precipitation and freeze and thaw dates, which can be used as a comparative for the climate change projections evaluated in this report.

Projected mean annual temperature, precipitation and freezing days all exhibit similar values. The NCEC report uses historical trends to predict the future outcomes. Coffee Gold's report uses global climate models, while this report uses both. Lorax (2015) in the Coffee Gold report uses five downscaled model averages climate change models. SRK's methodology does not use downscaled climatic model models, instead considered gridded information for all the available climate change models and compared those results applied to the site with historical trends from Reanalysis ERA-Interim.

The expected mean annual temperature presents similar trend in all of three reports. Freezing-day estimations are also quite similar in Coffee Gold's report and to SRK's. MAP shows an increasing trend in all reports; SRK's report presents a greater increase than either of the two benchmark reports. However, the magnitude seems reasonable compared with the benchmarked reports. Table 4.1 presents an overall synthesis of the climate change results in the area.

Table 4.1: Climate Change Benchmarks – Meteorological Parameters

Projected Parameter	Period [years]	NCEC Yukon, 2015	Lorax -Coffee Gold, 2015	SRK – Minto, 2016
Mean Annual Temperature	2060's	>+2 ˚C		+2.4 ˚C[1]
	2100's		+3-5 ˚C	+3.3 ˚C
Mean Annual Precipitation	2060's	+10-20%		+ 29% [1]
	2100's		+20%	+44%
Freezing Days	2060's	N/A		-18.5 Days[1]
	2100's		-30 Days	-29 Days

Note(s):

(1) Value defined for 2050s.

For IDF benchmarking the site IDF was compared with Simonovic et al's (Western University – London, ON) IDF CC tool (2016). They provide a web evaluation of all the IDF curves in Canada under different time frames and for a wide variety of climate change scenarios. The IDF information from the closest EC meteorological station of Pelly Ranch located 25 km north of the site was used from this web page as a comparison to the site IDF.

The climate change timeframe used in all of the scenarios is 2071 to 2100. Simonovic et al. (2016) provide IDF values for three representative concentration pathway (RCP) scenarios: RCP 2.6, RCP 4.5 and RCP 8.5. Table 4.2 provides a summary of the IDF changes for the Site (SRK) versus Pelly Ranch (Simonovic et. al., 2016) based on baseline conditions. The table shows that the SRK value falls within the higher range of the values presented by Simonovic et al. However, the value is within a reasonable range considering the uncertainty associated to these climate change models.

Table 4.2: Climate Change Benchmarks – Intensity Duration Frequency Curves

Source	RCP 2.6 (2071-2100)	RCP 4.5 (2071-2100)	RCP 8.5 (2071-2100)	Compiled (2071-2100)
IDF CC Tool – Pelly Ranch, 2016	+20%	+26%	+47%	
SRK – Minto, 2016				+44%

5 Summary of Overall Climate Change Trends

Climate change effects develop over very long timescales relative to the project life. As such the trends listed in Table 5.1 will manifest as minimal increases and decrease over the length of the Project.

Table 5.1: Overall Climate Change Trends

Climate Factor	Trend	Justification
Mean Annual Temperature	Increasing	The mean annual temperature should increase in total 3.3 degrees Celsius for the period 2100's.
Frequency of Extreme Temperatures	Reduction of variability	The intra-annual temperature range will decrease with the time, range between extreme maximum and minimum temperatures.
Frequency of Rainfall	Increasing	The IDF will increase in 44% for the period 2100s. This is also complemented with an increase in the amount of wet days in a year.
Total Rainfall	Increasing	Increase in total precipitation of 199.3 mm (44%) by 2100's.
Snow Accumulation	Increasing	Snow depth appears to be increasing between December and March mainly by years 2100s.
Snowmelt	Unknown	There is a predicted increase in the daily maximum snowmelt, however the values are not monotonically increasing with the time; and the variation magnitude is close to 10%, which suggest some uncertainty.
Floods and Storms	Unknown, likely Increasing	Increasing temperature, precipitation, long duration rainfall and storm intensities in the trends to 2100 suggest that storms and freshet are increasing. Therefore it is likely reasonable to assume that storms and floods will increase by the period of 2100's.

6 Effects of Climate Change on Closure Design Criteria

The climate change projections discussed in this memo and specifically summarized in Section 5 contribute to develop a long-term design criteria for engineered closure designs and infrastructure. For closure conveyance channel designs the predicted increase in precipitation and in the frequency of storms play a considerable role in engineering channel geometries and riprap specifications. According to the climate change study, channels which are designed on the present day discharges are projected to more water in the form of a greater peak flow, greater and more frequent storm runoffs, and even freshets flow.

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Appendix D – Channel Routing Options Evaluation

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Memo

To:	Ryan Herbert, Minto Explorations Ltd	**Client:**	Minto Explorations Ltd.
From:	Samantha Barnes, SRK	**Project No:**	1CM002.042
Cc:	Dylan McGregor, SRK	**Date:**	August 3, 2016
Subject:	Minto Channel Routing Options Evaluation		

1 Introduction

This memo summarizes the impacts of routing runoff through the pit versus diverting it around the pit. The Minto routing flow closure evaluation assessed channel design options for surface water flow received by the Main and Area 2 open pits.

Flow routing alternatives include two options for the Area 2 Pit:

- Option 1 – Area 2 Pit upstream flow are routed into Area 2 Pit through Channel B (Ditch B Phase V/VI Design). From the pit, the flows are discharged through channel width, which becomes Channel D.
- Option 2 – Area 2 Pit catchment is diverted into Channel B (Ditch B), which by-passes the Area 2 Pit.

Three 24 hours precipitation return period models were applied to three geometric channel configurations for the Main Pit and options 1 and 2 for the Area 2 Pit.

2 Hydraulic – Hydrologic Model

Three 24 hours precipitation return period events were modelled to test the change in resultant peak flow and flow depth in the discharge channel design: a 2-year peak flow, a 200-year peak flow, and a probable maximum flood (PMF). Three geometries were assessed for discharge channels from both open pits as follows:

- Case 1 – Minimum channel width (2 m)
- Case 2 – Constructible channel width, based on dozer sizing (7 m)
- Case 3 – Maximum channel width, based on topography (22 m for Main Pit, 15 m for Area 2 Pit)

Geometric models of each routing alternative and each channel width alternative were compared using HEC-HMS developed by US Army Corps of Engineers (2000).

3 Results

3.1 Routing Options – Hydrologic Results

Figure 1 shows the complete hydrograph for Option 1 and Option 2. The blue solid line is the routing and channel width alternative (Option 1). The red dashed line is the diversion alternative (Option 2). Figure 2 shows a close-up of this variation in peak flow under both scenarios. The difference between the two routing options is approximately 30 m³/s for the PMF event.

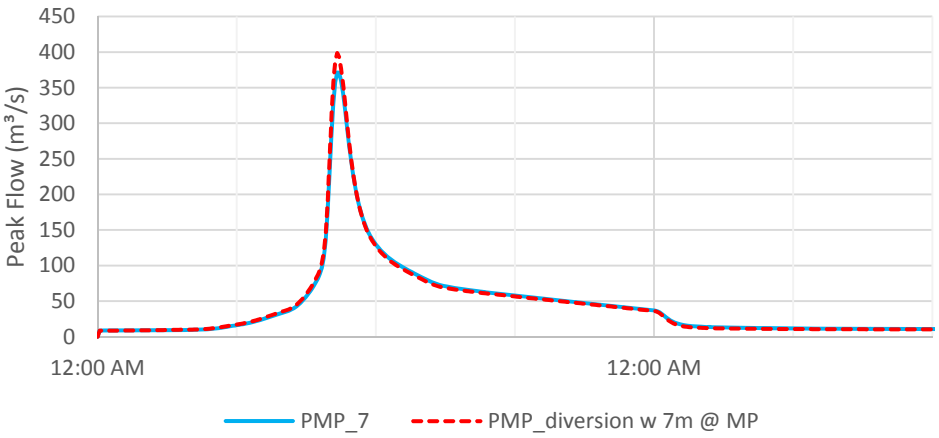


Figure 1: Routing Option Comparison for the PMF

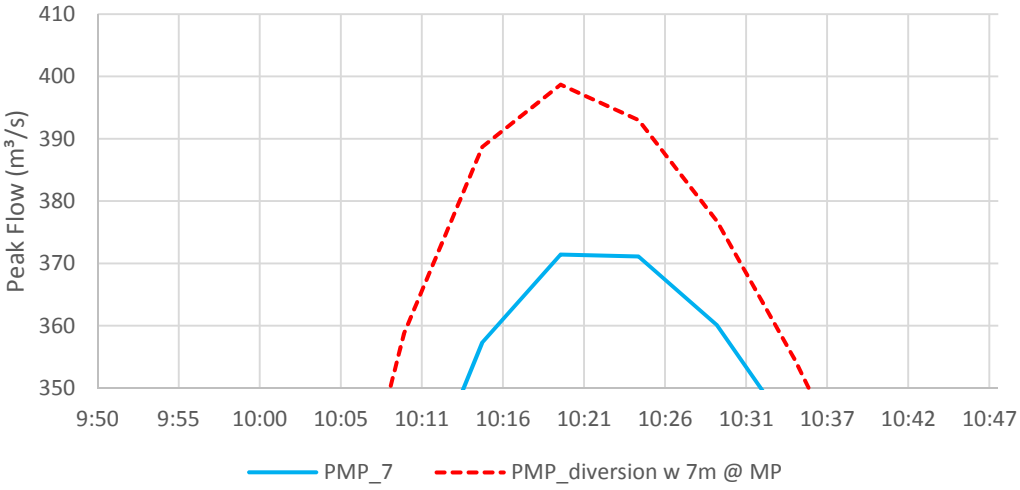


Figure 2: Routing Option Comparison for the PMF (Flow Peak Close-up)

3.2 Geometric Cases – Hydraulic Results

The results for the three channel width cases are presented in Table 1 where the design peak flow events are assumed to be the PMP. The riprap was sized based on the Isbash method (Garcia 2007).

Table 1: Probable Maximum Flood Results

Channel Width Scenario	Area 2 Pit			Main Pit		
	Peak Flow (m^3/s)	Flow Depth (m)	Riprap D50 (m)	Peak Flow (m^3/s)	Flow Depth (m)	Riprap D50 (m)
Case 1 (2m)	5.8	0.99	0.08	17.4	1.71	0.30
Case 2 (7m)	9.7	0.79	0.09	23	1.31	0.32
Case 3 (max)	14	0.64	0.07	31.9	0.86	0.17

4 Discussion and Recommendations

The results of the two routing options illustrate a minimal difference in impact from diverting runoff around the Area 2 Pit as opposed to through the pit. Furthermore, the diversion Option 2 does not result in a difference in the magnitude of peak flow or the period of time when the peak is presented when compared to routing flows through the pit. For the channel width options to minimize the size of riprap required, Case 2 is recommended for both open pit channel widths.

Based on the analyses presented and considering future construction riprap needs, it is recommended that the Option 1 channel alignment and the Case 2 channel width option be implemented. Option 1 represents the routine option with the lowest flows and it is paired with the channel geometry that requires the most conservative riprap sizing.

SRK Consulting (Canada) Inc.

Original Signed By

Samantha Barnes, EIT
Consultant

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Appendix E – Integrating Vegetation Erosion Protection

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Memo

To:	Ryan Herbert, Minto Explorations Ltd.	**Client:**	Minto Explorations Ltd.
From:	Stuart McPhee, EIT	**Project No:**	1CM002.042
Cc:	Dylan MacGregor, SRK	**Date:**	August 3, 2016
Subject:	Integrating Vegetation Erosion Protection		

This memo discusses the use of vegetation to assist channel stabilization and erosion reinforcement of disturbed soils for the Minto Project. Best management practices (BMPs) are provided in the Yukon Revegetation Manual for Practical Approaches and Methods (Matheus 2013) and the Government of Alberta's Transportation Erosion and Sediment Control Manual (2011). These BMP's provide guidelines for analysis, design selection, construction, and maintenance of erosion, and sediment control structures and site-specific selection of vegetation and establishing plant communities.

A commonly employed reclamation practice is to use live vegetation for channel protection and erosion control of disturbed soils. The use of biotechnical techniques do not replace the need for rock riprap, but can complement and improve channel stabilization designs during construction (e.g. regraded slopes and rock riprap). These techniques also improve soil reinforcement on disturbed slopes and regraded areas and ultimately assist in establishing vegetation, riparian zones, and develop soils.

The vegetation proposed for channel stabilization and erosion control focus on the use of coarse woody, grass and legume, and wetland species of vegetation. These species are widely used in industrial site reclamation design, construction, and maintenance to mitigate erosion, and sedimentation. The Yukon Revegetation Manual summarizes common plant species that are established by seeding. Common seeding practices for reclamation include hydroseeding and broadcast seeding. Appended to this memo are the summary tables prepared in the Yukon Revegetation Manual *Chapter 5: Revegetation Scenarios and Seed Recommendations*. Included in the appended tables are both Non-Native (Agronomical) and Locally-Collected (Native) seeds and the reclamation scenario's that they are suited for.

Methods to improve existing and proposed infrastructure (e.g. conveyance channels, covers and disturbed erodible slopes) include:

- establish vegetation and restorative self-sustaining plant communities;
- increase slope and channel stability due to root reinforcement, interlock, and soil draining;
- provide protection against wind, water, and surface erosion and rock fall;

- provide shade to regulate temperature and humidity close to surface that promotes further vegetation growth of volunteer species;
- provide improvement to the soil water regime through interception, evapotranspiration, and storage;
- improve soil development and the formation of organic soils;
- improve habitat provisions for animals;
- may reduce construction and maintenance costs (compared to other designs); and
- establish post closure landscapes that are often more appealing due to their more natural features.

At this stage of closure design, a preliminary assessment of common biotechnical reclamation practices has identified that there is a future opportunity to integrate vegetation erosion protection into the closure conveyance channel design and the disturbed soils adjacent to the channels.

There are three main components to the design and construction of the closure conveyance channels that would benefit from biotechnical stabilization BMPs:

- Disturbed and excavated slopes
- Establishing riparian vegetation
- Channel erosion and protection

Closure designs for conveyance channels will require sections of channels to have excavated alignments through erodible materials. The excavated slopes within these materials will typically have slopes graded towards the channel. It is expected that if these disturbed slopes are left, they will erode and even transport sediment into the channels. If the slopes contain weak fine grained erodible soils, then the slopes may be subject to sloughing and larger scale transportation of materials, which may block the conveyance channels.

Immediately after construction the closure conveyance channels will lack riparian vegetation. Biotechnical techniques can assist with slope stabilization and establishing riparian vegetation. Techniques for stabilizing these slopes are recommended to be a combination of BMPs provided in the Government of Alberta's Transportation Erosion and Sediment Control Manual (2011) as listed in Table 1. The BMP figure for live staking method is attached to this memo as an example.

Table 1: Selected Closure BMPs from Alberta Transportation Erosion and Sediment Control Manual

BMP ID	BMP Name	Disturbed and Excavated Slopes	Riparian Vegetation Development	Channel Erosion Protection
13b	Rolled Erosion Control Products	✓		
14a,b	Riprap Armouring	✓		✓
19a,b	Slope Drains	✓	✓	
24a	Hydroseeding	✓	✓	

BMP ID	BMP Name	Disturbed and Excavated Slopes	Riparian Vegetation Development	Channel Erosion Protection
27a	Live Staking	✓	✓	✓
27b1,2,3	Brush Layering	✓	✓	✓
28	Wattle (Live Fascine)	✓	✓	
34a	Surface Roughening	✓	✓	
34c	Slope Benching	✓		
38a,b,c	Coir/Straw Rolls with Brush Layers	✓	✓	
39	Brush Mattress	✓	✓	
40	Live Siltation		✓	✓
41	Willow Post and Pole	✓	✓	✓
44	Vegetated MSE	✓	✓	✓
45a,b,c,d	Vegetated Riprap	✓	✓	✓

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Attachment 1: Revegetation Scenarios and Seed Recommendations

Summary of Recommended Species

Table 5.27

NON-NATIVE HERBACEOUS SPECIES (COMMERCIAL SEEDS) AND APPLICABLE SCENARIOS

SCENARIO		1	2	3	4	5	6	7
SPECIES (cultivar)	**# OF SEEDS/KG**	LOW SLOPE, LOW-MID ELEVATION SITES	STEEP SLOPES	WET SITES	HIGHWAY RIGHT-OF-WAYS	DISTURBED PERMAFROST SITES	ARCTIC TUNDRA AND STABLE PERMAFROST SITES	ALPINE AND SUBALPINE SITES
Rocky Mountain fescue *Festuca saximontana* (common)	**1,430,000**	x	x		x	x		x
Violet wheatgrass *Elymus alaskanus* (common)	**330,000**	x	x		x		x	x
Glaucous bluegrass *Poa glauca* (Tundra or common)	**2,904,000**	x	x		x		x	x
Slender wheatgrass *Elymus trachycaulus* (Revenue, Adanac, Highlander or common)	**349,800**	x	x		x			
Ticklegrass *Agrostis scabra* (common)	**11,000,000**	x				x		x
Tufted hairgrass *Deschampsia caespitosa* (Nortran or common)	**3,300,000**	x		x	x	x	x	x
Fowl bluegrass *Poa palustris* (common)	**4,840,000**	x		x	x	x		
Spike trisetum *Trisetum spicatum* (common)	**5,500,000**						x	x
Bluejoint reedgrass *Calamagrostis Canadensis* (Sourdough or common)	**8,442,438**	x		x		x		
Polargrass *Arctagrostis latifolia* (Alyeska or common)	**3,960,000**			x		x	x	
American sloughgrass *Beckmannia syzigachne* (Egan or common)	**2,723,600**			x		x		
Alpine bluegrass *Poa alpina* (Gruening or common)	**2,353,826**						x	x

Summary of Recommended Species

Table 5.28

NON-NATIVE HERBACEOUS SPECIES (COMMERCIAL SEEDS) AND APPLICABLE SCENARIOS

SCENARIO		1	2	3	4	5	6	7
SPECIES (cultivar)	**# OF SEEDS/KG**	LOW SLOPE, LOW-MID ELEVATION SITES	STEEP SLOPES	WET SITES	HIGHWAY RIGHT-OF-WAYS	DISTURBED LOWLAND PERMAFROST SITES	ARCTIC TUNDRA AND STABLE PERMAFROST SITES	ALPINE AND SUBALPINE SITES
Kentucky bluegrass *Poa pratensis* (Nugget)	**3,057,648**			x			x	
Creeping red fescue *Festuca rubra* (Arctared for northern Yukon, Boreal for southern Yukon)	**539,407**	x	x		x	x	x	x
Canada bluegrass *Poa compressa* (Reubens or common)	**5,264,204**	x	x	x	x	x		
Streambank wheatgrass *Elymus lanceolatus* (Sodar)	**336,600**	x	x	x	x	x	x	
Meadow foxtail *Alopecurus pratensis* (common)	**895,136**	x	x	x	x	x	x	
Red top *Agrostis gigantea* (common)	**10,672,640**	x		x				
Timothy *Phleum pratense* (Climax, Engmo)	**2,559,400**			x		x	x	
Alfalfa *Medicago sativa* (Rangelander, Rambler or Peace)	**498,960**		x					
Sheep fescue *Festuca ovina* (common)	**1,100,000**	x			x	x	x	x
Annual rye *Lolium multiflorum*	**477,400**		x			x		
Barley *Hordeum vulgare*	**29,920**		x			x	x	x

Summary of Recommended Species

Table 5.29

LOCALLY-COLLECTED NATIVE HERBACEOUS SPECIES (SEEDS) AND APPLICABLE SCENARIO

SCENARIO		1	2	3	4	5	6	7
SPECIES	# OF SEEDS/KG	LOW SLOPE, LOW-MID ELEVATION SITES	STEEP SLOPES	WET SITES	HIGHWAY RIGHT-OF-WAYS	DISTURBED LOWLAND PERMAFROST SITES	ARCTIC TUNDRA AND STABLE PERMAFROST SITES	ALPINE AND SUBALPINE SITES
Northern rough fescue *Festuca altaica*	451,854	x	x	x	x	x	x	
Yukon wheatgrass *Elymus calderi*	—	x						
Macrourum's wheatgrass *Elymus macrourus*	374,782	x	x					
Northern brome *Bromus pumpellianus*	202,400	x	x		x			
Sweetgrass *Hierochloë hirta* (formerly *H. odorata*)	242,000	x	x		x			
Alpine sweetgrass *Hierochloë alpina*	—						x	x
Mountain timothy *Phleum alpinum* (formerly *P. commutatum*)	—					x		x
Northern bluegrass *Poa alpigena/Poa pratensis*	3,057,648			x		x	x	x
Arctic bluegrass *Poa arctica*	3,250,553						x	
Nuttall's alkaligrass *Puccinellia nuttalliana*	4,648,140	x	x	x	x	x	x	x
Yellow locoweed *Oxytropis campestris*	521,400	x			x			
Showy locoweed *Oxytropis splendens*	1,548,800	x			x			
Bear root *Hedysarum alpinum*	154,000	x						
Mackenzie's hedysarum *Hedysarum Mackenzii*	101,889	x						
Yarrow *Achillea millefolium* and *Achillea sibirica*	6,274,426	x			x			
Mountain avens *Dryas* spp.	—	x			x	x	x	x
Arctic lupine *Lupinus arcticus*	22,733	x		x	x	x	x	x
Wormwood/sage *Artemesia* spp.	—	x	x		x		x	x

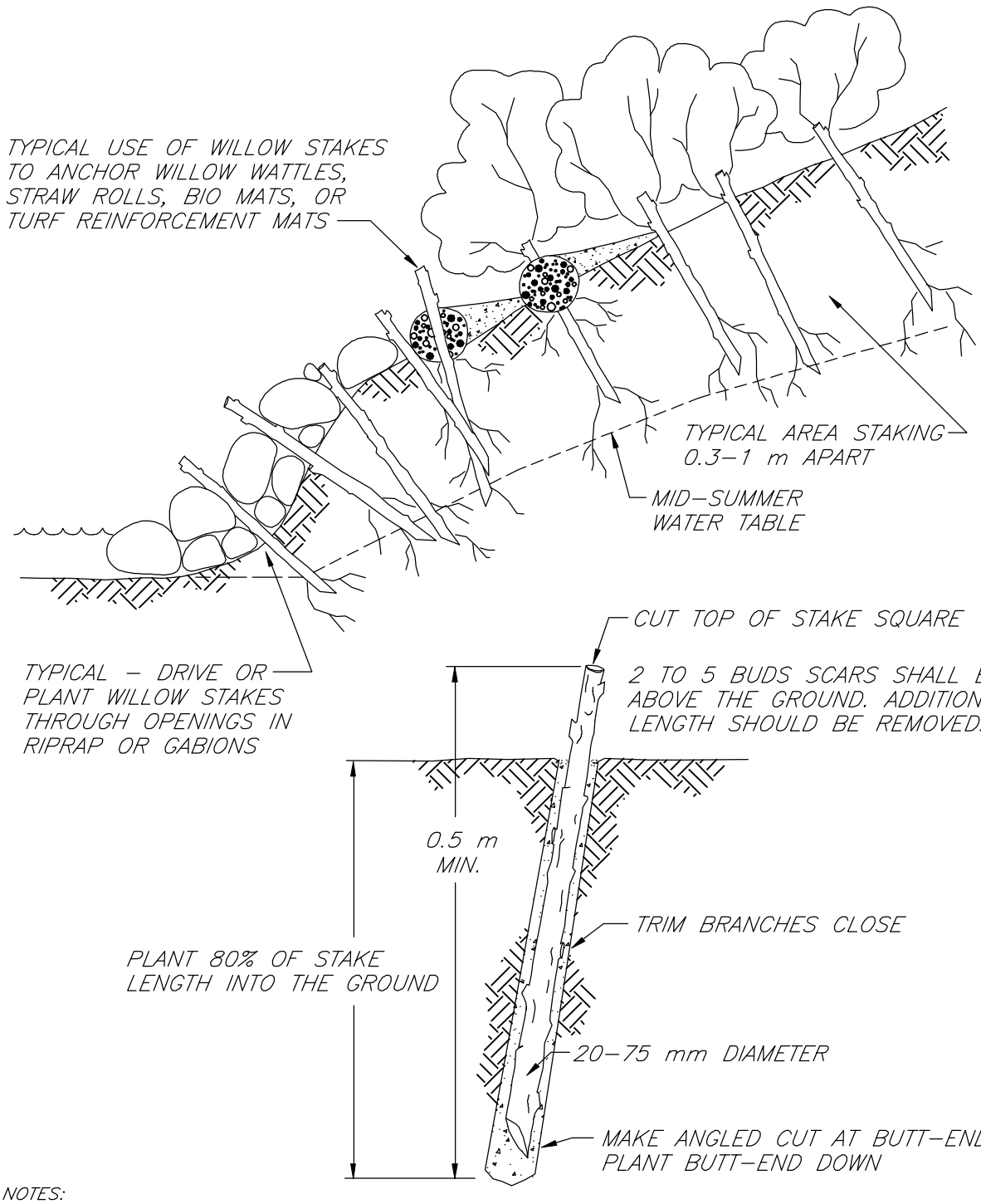
Summary of Recommended Species

Table 5.30

LOCALLY-COLLECTED WOODY SPECIES AND APPLICABLE SCENARIOS

SCENARIO	1	2	3	4	5	6	7
LOCAL WOODY SPECIES	LOW SLOPE, LOW-MID ELEVATION SITES	STEEP SLOPES	WET SITES	HIGHWAY RIGHT-OF-WAYS	DISTURBED LOWLAND PERMAFROST SITES	ARCTIC TUNDRA AND STABLE PERMAFROST SITES	ALPINE AND SUBALPINE SITES
Willow *Salix alaxensis, S. pulchra, S. planifolia, S. richardsonii, S. hastate, S. arbusuloides, S. lucida, S. pseudomyrsinites* (the two best species are *S. alaxensis* and *S. pulchra*)	x		x		x	x	x
Poplar *Populus balsamifera*	x		x				
Shrub birch *Betula glandulosa*	x		x			x	x
Dwarf birch *Betula nana*						x	
Green alder *Alnus crispa*	x		x			x	x
Grey alder *Alnus tenuifolia*	x		x				

Attachment 2: Live Staking BMP Technique



NOTES:
1. HARVEST AND PLANT STAKES DURING THE DORMANT SEASON.
2. USE HEALTHY, STRAIGHT AND LIVE WOOD AT LEAST 1 YEAR OLD.
3. MAKE CLEAN CUTS AND DO NOT DAMAGE STAKES OR SPLIT ENDS DURING INSTALLATION, USE A PILOT BAR IN FIRM SOILS.
4. SOAK CUTTINGS FOR 24 HOURS (MIN.) PRIOR TO INSTALLATION.
5. TAMP THE SOIL AROUND THE STAKE.
6. THIS FIGURE IS PROVIDED FOR GUIDANCE ONLY AND DOES NOT CONSTITUTE A DESIGN. A SITE SPECIFIC DESIGN IS REQUIRED FROM DESIGNER/ENGINEER.

NOT TO SCALE

LIVE STAKING

FILE: LIVESTK

From: Salix–Applied Earthcare – EROSION DRAW 3.0
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Site Characterization Plan

January 2018

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1 INTRODUCTION

As a requirement for Capstone Mining Corp's Quartz Mining Licence, a Site Characterization Plan has been developed. The Plan will outline the existing condition at the Minto Mine. Table 1-1 shows how sections of this plan correspond to the Plan Requirements Guidance for Quartz Mining Projects.

Table 1-1: Concordance with Plan Requirements

Section in Guidance Document		Corresponding Section(s) in this Plan	
4.1	Introduction	1	Introduction
4.2	Geology	2	Geology
4.3	Climate and Hydrology	3 4	Meteorology Hydrology
4.4	Surface Water	5	Surface Water Quality
4.5	Groundwater	6	Groundwater
4.6	Vegetation and Wildlife	7 8 9	Wildlife Aquatic Resources Vegetation
4.7	Soils and Bedrock	2	Geology
4.8	Seismicity	2	Geology
4.9	Geochemistry and Geotechnical Information	10	Geochemistry and Geotechnical Information

The Plan Requirements Guidance for Quartz Mining Projects references a concordance table outlining applicable proponent commitments and decision document conditions. As this plan describes the existing site conditions, none of the commitments made in the YESAB process or the conditions in the decision document are relevant to the understanding of this plan. Capstone's commitments and the conditions contained in the decision document are addressed in the appropriate documents that evaluate the potential environmental effects of the project and the issues raised in the evaluation process.

1.1 SITE OVERVIEW

The Minto Mine is located on the west side of the Yukon River, approximately 240 km northwest of Whitehorse, Yukon (Figure 1-1) and is centered at 62°37'N latitude and 137°15'W longitude (NAD 83, UTM Zone 8 coordinates 6945000N, 384000E). Highway 2 (North Klondike Highway) is located on the east side of the Yukon River; the mine can be accessed in the summer by barge crossing or in winter by the ice bridge crossing at Minto Landing.

Environmental conditions are summarized in Table 1-2 below.

Table 1-2: Summary of Environmental Conditions in the Minto Mine Area.

Project Area Attribute Description	Project Area Attribute Description
Region:	Yukon
Topographic map sheet:	NTS 115 I/10, 115 I/11
Geographic location name code:	Minto Project
Latitude:	62° 36' N
Longitude:	137° 15' W
Drainage region:	Yukon River
Watersheds:	Yukon River, Big Creek, Wolverine Creek, Dark Creek, McGinty Creek, and Minto Creek.
Ecoregion:	Yukon Plateau (Central) - Pelly River ecoregion.
Study area elevation:	Rolling hills above mine site at 1131 to 600 m at the Yukon River Valley bottom.
Site climate:	Temperature ranges from –43.2°C (November 2006) to 30.3°C (July 2009). Mean annual temperature of –1.8°C. Mean annual rainfall is 174mm.
Vegetation communities:	Riparian, black spruce, white spruce, paper birch, lodgepole pine, buck brush/willow and ericaceous shrubs, feather moss, sedge, sagewort, grassland, mixed forest (aspen, balsam, and sub-alpine). Discontinuous permafrost is present on site. Site has been subject to recent forest fires.
Wildlife species:	Moose, caribou, Dall sheep, mule deer, grizzly and black bear, varying hare, beaver, lynx, marten, ermine, deer mouse, fox, mink, wolverine, least weasel, wolf, squirrel, porcupine coyote, muskrat, otter and wood frog. Bird species include: spruce, blue, ruffed, and sharp-tail grouse, waterfowl, raptors, and a variety of smaller birds.
Fish species:	In the Yukon River, Chinook, Coho, and chum salmon, rainbow trout, lake trout, least cisco, Bering cisco, round whitefish, lake whitefish, inconnu, Arctic grayling, northern pike, burbot, longnose sucker and slimy sculpin; In Big Creek, Chinook and chum salmon, Arctic grayling and whitefish species; In Wolverine Creek, Chinook salmon, Arctic grayling, and slimy sculpin; In Minto Creek (lower reaches only), Chinook salmon, slimy sculpin, round whitefish, Arctic grayling, longnose sucker, burbot; In McGinty Creek (lower reaches only), slimy sculpin, Arctic grayling

The Minto Mine property lies in the eastern portion of the Dawson Range, which is part of the Klondike Plateau Physiographic Region, an uplifted surface that has been dissected by erosion. The area was largely unglaciated during the last ice age and topography consists of deep and narrow valleys, rounded rolling hills, and ridges with relief of up to 600 m (2,000 ft.). The highest elevation on the property is 975 m (3,200 ft.) above sea level, compared to lower elevations in the region with elevations of 460 m (1,500 ft.) along the Yukon River.

The Minto Mine is near the height of land with relatively gentle slopes and smooth ridges that often have spines of bedrock outcrops (tors) at their crests left from long periods of weathering. Broad ridges are typically mantled with felsenmeer (fields of angular, frost-heaved, in situ rock fragments). Below the ridge crests, bedrock exposures are limited or negligible. The Project area lies in a zone of extensive discontinuous (50–90%) permafrost and is included in the western portion of the Yukon plateau central ecozone. North-facing slopes in the area are commonly underlain by permafrost. Valley-bottom deposits and upland soils usually contain ice-rich horizons. Well-drained uplands may have permafrost-free soils.

As the ecoregion is largely unglaciated, the dominant parent materials are stony residual materials along ridge tops and summits, coarse colluvium on upper slopes, and silty colluvium and loess, rich in organic matter, on lower slopes and floors of main valleys. Muck is usually capped with peat and underlain by

permafrost. In the Minto Mine area, much of the colluvium on ridge slopes is coarse sand derived from decomposition of the largely granitic bedrock in the area. Overburden of relic soils that predate the last glaciation includes lacustrine deposits overlain by colluvium and organic silt in some upper valleys, some of which was exposed during excavation on the south side of the pit.

The mine development area is in the transition from the forested to non-forested (alpine) zone. Below the treeline (elev. ~1,000 m) the vegetation patterns reflect the discontinuous distribution of permafrost with stunted black spruce woodlands on cold, north-facing sites and mixed (aspen, white spruce, minor birch) forests on warm, south-facing slopes. The area has been burned over by several wild fires, the latest of which was in 2010 (Minto Landing to lower Minto Creek area). Many of the burnt trees have blown down and natural regeneration of pine and alder is occurring over much of the property and the Project area.

The climate of the region is continental with short, warm summers and very cold winters. Annual precipitation ranges from 300 to 500 mm. Mean annual temperatures are near –5°C with mean mid-winter temperatures of –23°C to –32°C, in July from +10°C to +15°C and extremes in the lower valleys ranging from –60°C to +35°C.

The area is drained by tributaries of the Yukon River. The rate of runoff is controlled by almost total vegetation ground cover and moderate slope gradients. Infiltration rate is expected to be high in areas of thicker colluvium. However, permafrost and seasonally frozen soils inhibit vertical percolation.

Topography in the Dawson Range is moderate and the active geomorphological processes in the study area are typically limited to include slow mass movement (solifluction) and some minor gully erosion. Although there are no reports of large-scale active natural landslides in the Minto Project area, smaller scale instabilities have recently been documented in the Minto Creek catchment area, downstream of the mine site. Substantial sediment loading in lower Minto Creek in the absence of effluent discharge from the site was observed throughout the open water months of 2011 (Appendix 1-3, *Minto Creek Water Quality Characterization*). This prompted a targeted water quality survey and an aerial investigation of the creek and tributaries in spring of 2012 in an attempt to identify the source(s) of the sediment. Two areas were identified in tributaries on the southern side of the Minto Creek catchment, below the area of control of the mine, where soil materials had slumped and transported a significant amount of material into the creek channel. Although ground investigations were not conducted, it is suspected that these instances may be related to thawing of frozen ground conditions, or may simply be small-scale instability of fine textured (lacustrine) soils.

1.2 SITE DESCRIPTION

The Minto Mine is a high grade copper mine located within Selkirk First Nation (SFN) Category A Settlement Land Parcel R-6A approximately 240 km northwest of Whitehorse, Yukon Territory (Figure 1-1). It is owned and operated by the Minto Explorations Ltd. a wholly owned subsidiary of Capstone Mining Corporation. Development of the mine was initiated in 1997, commercial operations started in October 2007, and the currently anticipated operating life is to 2020 with permitted resources to 2022. The Minto Mine is permitted

to conduct open pit and underground mining, and to mill the copper/gold/silver ore at a rate of 4,200 tonnes of per day. In 2016 (the most recent year of record), the Minto Mine produced 69.3 million pounds of copper from milling approximately 1.49 million tones or ore.

In addition to open pit mines, underground mines, and the mill, the Minto Mine site includes a number of waste rock dumps, a concentrate storage shed, a dry stack tailings storage facility (DSTSF), a water retention dam with a water storage pond (WSP), a water treatment plant, administrative offices, an airstrip, and a camp (Figure 1-2). Mill tailings are stored in the DSTSF and in mined-out open pits. Mine-impacted seepage from the DSTSF and under the Mill Valley Fill (MVF) is collected at the toe of the MVF in sump W62 and is pumped to the Main Pit (Figure 1-2). Non-impacted water and treated mine-impacted water are collected in the WSP (Figure 1-2). Effluent from the WSP is periodically discharged to Minto Creek under conditions specified in Water Use Licence (WUL) QZ14-031 (August 2015). Minto Creek, in turn, discharges to the Yukon River approximately 7.7 km south-east of the WSP (Figure 1-2).





capstone
MINTO MINE

ALEXCO ENVIRONMENTAL GROUP

MINTO MINE
CAPSTONE MINTO MINE - SITE CHARACTERIZATION PLAN

FIGURE 1-1
PROJECT LOCATION

JANUARY 2018

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MINTO MINE

CAPSTONE MINTO MINE - SITE CHARACTERIZATION PLAN

FIGURE 1-2

MINE SITE LAYOUT

JANUARY 2018

capstone MINTO MINE

1:13,500 *when printed on 11 x 17 inch paper*

0 50 100 200 300 400 500 Meters

ALEXCO ENVIRONMENTAL GROUP

Aerial imagery obtained from Challenger Geomatics. Imagery acquired August 9rd 2017

Datum: NAD 83 Projection: UTM Zone 8N

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2 GEOLOGY

2.1 PHYSIOGRAPHY

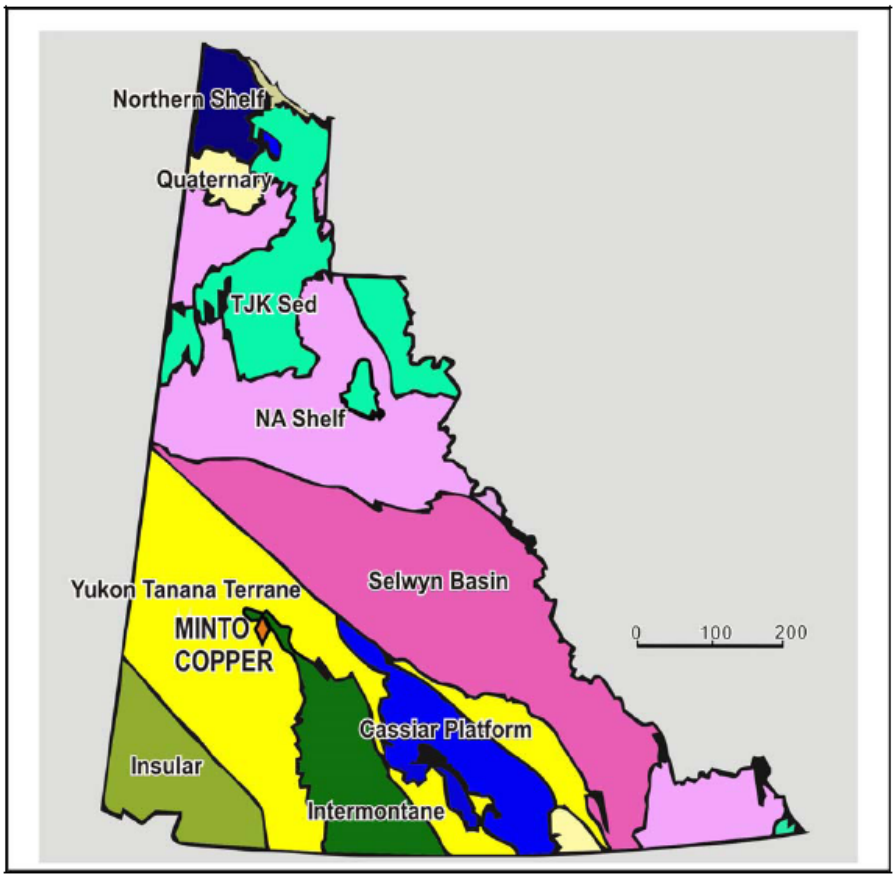
The property lies in the Dawson Range, part of the Klondike Plateau, an uplifted surface that has been dissected by erosion. Local topography consists of rounded rolling hills and ridges and broad valleys. The highest elevation on the property is approximately 1,000 m above sea level, compared to elevations of 460 m along the Yukon River. Slopes on the property are relatively gentle and do not present accessibility problems. Bedrock outcrops can often be found at the tops of hills and ridges. There are no risks of avalanche on the property.

Overburden is colluvium primarily comprised of granite-based sand from weathering of the granitic bedrock in the area and is generally thin but pervasive. It can reach plus 50 m in depth. Seams of clay and ice lenses are also present sporadically. South-facing slopes generally provide well-drained, sound foundation for buildings and roads. North-facing slopes in the area typically contain permafrost.

Vegetation in the area is sub-Arctic boreal forest made up of largely spruce and poplar trees. The area has experienced several wildfires over the years, the latest in 2010, and has no old-growth trees remaining. The fire in 2010 led to the partial evacuation of the camp and a short stoppage in production.

2.2 BEDROCK GEOLOGY

The Minto Project is found in the north-northwest trending Carmacks Copper Belt along the eastern margin of the Yukon-Tanana Composite Terrain, which is comprised of several metamorphic assemblages and batholiths (). The Belt is host to several intrusion-related Cu-Au mineralized hydrothermal systems. The Yukon-Tanana Composite Terrain is the easternmost and largest of the pericratonic terrains accreted to the Paleozoic northwestern margin of North America (e.g., Colpron *et al.*, 2005). It is regarded to be the product of a continental arc and back-arc system, preserving meta-igneous and metasedimentary rocks of Permian age on top of a pre-Late Devonian metasedimentary basement (e.g., Piercey *et al*. 2002).



From: Yukon Geologic Survey "Maps Yukon" website (www.geology.gov.yk.ca)

Figure 2-1: Yukon Geology (from Yukon Geologic Survey "Maps Yukon" website (www.geology.gov.yk.ca).

The Minto Property and surrounding area are underlain by plutonic rocks of the Granite Mountain Batholith (Early Mesozoic Age) (Figure 2-2: Regional Geology) that have intruded into the Yukon-Tanana Composite Terrain. They vary in composition from quartz diorite and granodiorite to quartz monzonite. The batholith is unconformably overlain by clastic sedimentary rocks thought to be the Tantalus Formation and andesitic to basaltic volcanic rocks of the Carmacks Group, both of which are assigned a Late Cretaceous age. Immediately flanking the Granite Mountain Batholith, to the east, is a package of undated mafic volcanic rocks, outcropping on the shores of the Yukon River. The structural relationship between the batholith and the undated mafic volcanics is poorly understood because the contact zone is not exposed

Geobarometry and geothermometry data (Tafti and Mortensen, 2004) suggests that the Granite Mountain Batholith was emplaced at a depth of at least 9 km, while the presence of euhedral to subhedral epidote, interpreted by Tafti and Mortensen as magmatic in origin, suggests a deeper emplacement depth in the order of 18 to 20 km.

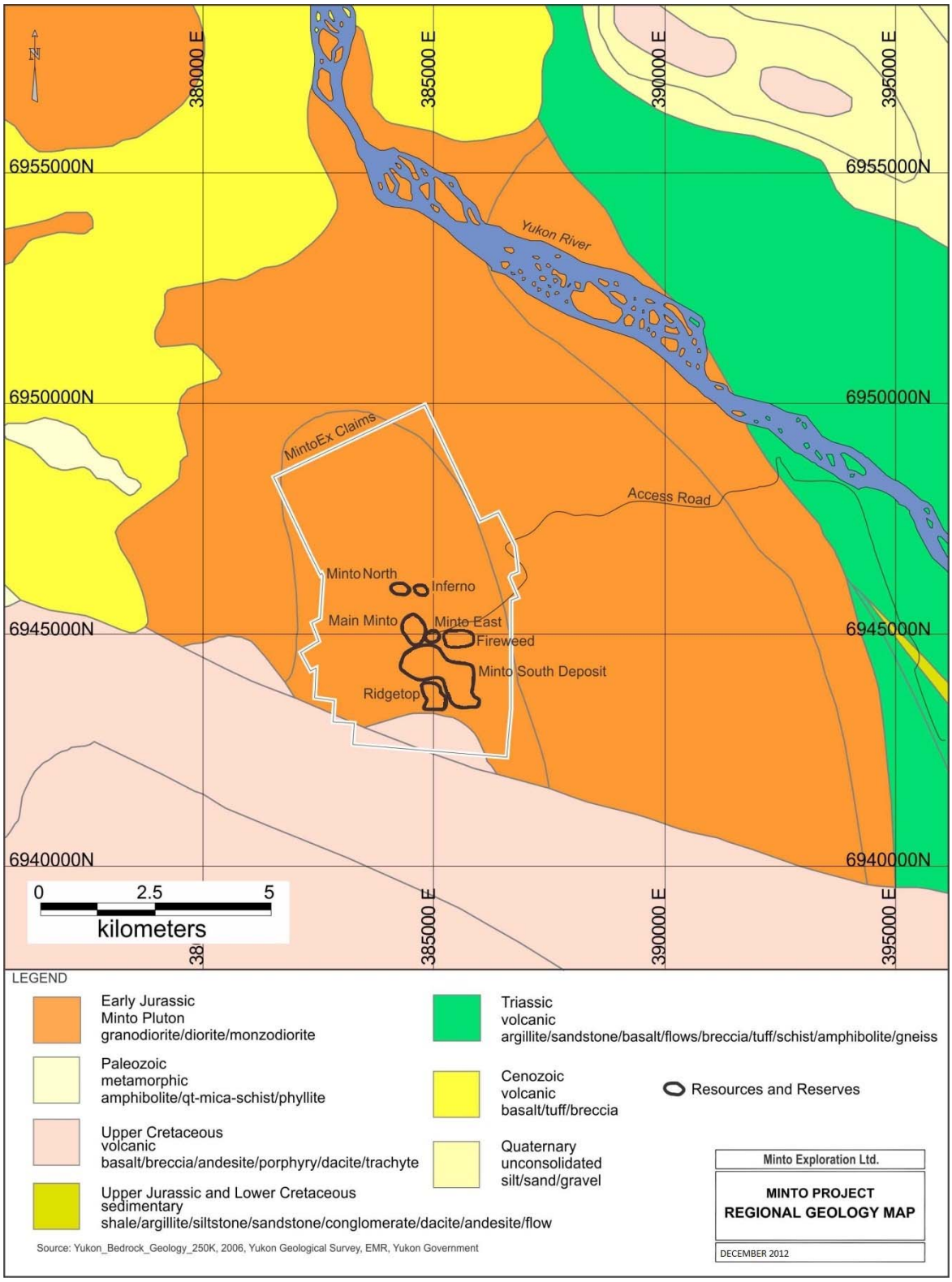


Figure 2-2: Regional Geology.

2.3 PROPERTY GEOLOGY AND LITHOLOGICAL DESCRIPTION

Much of the geological understanding of the rock around the Minto deposits is based on observations from diamond drill core and extrapolation from regional observations. The reason for this is poor outcrop exposure (less than 5% coverage), as well as the deep weathering and oxidation of any existing exposed outcrop. The terrain was not glaciated during the last ice age event.

Five additional deposits of mineralization are reported in this document outside of the Minto Main Deposit; Minto South Deposit, Ridgetop, Minto North, Inferno, Fireweed, and Minto East (). For the purpose of this report Area 2, Area 118, Copper Keel, and Wildfire resource sub-domains are now considered continuous and reported as one deposit, namely Minto South Deposit (MSD) located immediately south of Main Minto. Names of these resource sub-domains are retained for ease of discussion when discussing regions within MSD. The Ridgetop deposit is located just over 300 m south of the Area 2/118 resource sub-domain, the Minto North and Inferno deposits are located about 700 m north of the Main Deposit, the Fireweed deposit is located just over 500m east of the south end of the Minto Main deposit, the Minto East deposit is located about 200 m east of the south end of the Minto Main deposit, while the Copper Keel and Wildfire sub-domains form a southeast extension from Area 2/118. Each of these deposits closely shares a similar style of mineralization of shallow dipping copper sulphide mineralized zones. The Main Minto deposit was exposed in the dormant open pit mine; similarly, this flat geometry is currently exposed in the Minto South Deposit (Area 2 resource sub-domain) starter pit. In addition to these mineral deposits which have NI 43-101 compliant mineral resources there are several significant mineral prospects. These deposits and prospects define a general north-northwest trend.

The hypogene copper sulphide mineralization at Minto is hosted wholly within the Minto pluton, which intrudes near the boundary between the Stikinia and Yukon-Tanana terrains, however since the contact is not exposed it is unclear if the pluton stitches the two terrains. The Minto pluton is predominantly of granodiorite composition. Hood *et al*. (2008) distinguish three varieties of the intrusive rocks in the pluton. The first variety is a megacrystic K-feldspar granodiorite. It gradually ranges in mineralogy to quartz diorite and rarely to quartz monzonite or granite, typically maintaining a massive igneous texture. An exception occurs locally where weakly to strongly foliated granodiorite is seen in distinct sub-parallel zones several meters to tens of meters thick.

A second variety of igneous rock is quartzofeldspathic gneiss with centimeter-thick compositional layering and folded by centimetre to decimetre-scale disharmonic, gentle to isoclinal folds (Hood et al., 2008). The third variety of intrusive is a biotite-rich gneiss. MintoEx geologists consider all units to be similar in origin and are variably deformed equivalents of the same intrusion.

Copper sulphide mineralization is found in the rocks that have a structurally imposed fabric, ranging from a weak foliation to strongly developed gneissic banding. For this reason, all core logging by the past and present operators separates the foliated to gneissic textured granodiorite as a distinctly discernible unit. It is generally believed by MintoEx geologists that the foliated granodiorite is just variably strained equivalents of the two primary granodiorite textures and not a separate lithology.

While this interpretation, based upon detailed observations from logging of tens of kilometers of drill core is highly likely but it still needs to be conclusively proven. Tafti & Mortensen (2004) noted that the relatively massive plutonic rocks have similar mineral and chemical composition as the foliated rocks. Research in collaboration with the Mineral Deposits Research Unit ("MDRU") of the University of British Columbia is on-going.

The contact relationship between the foliated deformation zones and the massive phases of granodiorite is generally very sharp. These contacts do not exhibit chilled margins and are considered by MintoEx geologists to be structural in nature, separating the variably strained equivalents of the same rock type. Tafti and Mortensen (2004) had interpreted the sharp contacts to be zones of deformed rock within the unfoliated rock (i.e. rafts or roof pendants). Supergene mineralization occurs proximal to near-surface extension of the primary mineralization and beneath the Cretaceous conglomerate.

Conglomerate and volcanic flows have been logged in drill core by past operators. New drilling has confirmed the presence of conglomerate, but not the volcanic flows. The latter cannot be confirmed by the authors as the drill core from historic campaigns was largely destroyed in forest fires and no new drilling has intersected such rocks. However, undated volcanic rocks are mapped by Hood, near the southwest margin of the property, south of a fault that is inferred from geophysics to separate them from the Jurassic Age intrusive rocks. The conglomerate has been dated (unpublished date pers. com. Dr. Maurice Colpron - Yukon Geological Survey) as Cretaceous Age. It is now recognized as an outcrop within a borrow pit exposure located west of the airstrip as well as in numerous recent drill holes. Observations of foliated and even copper mineralized cobbles in drilling indicate that "Minto-type" mineralization was exposed, eroded and reincorporated in sedimentary deposits by the Cretaceous Age.

Other rock types, albeit volumetrically insignificant, include dykes of simple quartz-feldspar pegmatite, aplite; and an aphanitic textured intermediate composition rock. Bodies of all of these units are relatively thin and rarely exceed the one meter core intersections. These dykes are relatively late, and observed contact relationships suggest they generally postdate the peak ductile deformation event; however some pegmatite and aplite bodies observed in a rock cut located north of the mill complex are openly folded.

It is unclear if this folding is contemporaneous with foliation development in the deformed rocks or post-dates the foliation development. Observations from drill core and open cut benches in the mine show examples where the foliation and the pegmatitic/aplitic intrusions are both folded, as well as examples where the intrusions are not folded, suggesting two populations of minor dykes.

2.4 SEISMIC HAZARD

The tectonics and seismicity of southwestern Yukon are influenced primarily by the Pacific and North American lithospheric plate margins. In Yukon's St. Elias region, northwest British Columbia and southeast Alaska, the boundary of the two lithospheric plates changes from right lateral transform to subductive. Instead of sliding past each other, the Pacific Plate is forced beneath the stable North American plate resulting in the St. Elias region being uplifted. This transfer of force along the fault into uplift or mountain

building dissipates tectonic energy, reducing seismic effects on the region northeast of and across the fault (SRK, 2013).

An assessment of peak ground acceleration was performed for the Minto project area using the 2015 National Building Code Seismic Hazard Calculation, and is shown in below. Additional details can be found in Appendix 1-1 Seismic Hazard Assessment for Minto, YT.

2015 National Building Code Seismic Hazard Calculation

INFORMATION: Eastern Canada English (613) 995-5548 français (613) 995-0600 Facsimile (613) 992-8836
Western Canada English (250) 363-6500 Facsimile (250) 363-6565

July 31, 2016

Site: 62.618 N, 137.241 W User File Reference: Minto Mine

Requested by: , Minto Explorations Ltd.

National Building Code ground motions: 2% probability of exceedance in 50 years (0.000404 per annum)

Sa(0.05)	Sa(0.1)	**Sa(0.2)**	Sa(0.3)	**Sa(0.5)**	**Sa(1.0)**	**Sa(2.0)**	**Sa(5.0)**	**Sa(10.0)**	PGA (g)	PGV (m/s)
0.154	0.229	**0.269**	0.246	**0.207**	**0.146**	**0.083**	**0.030**	**0.012**	**0.125**	**0.156**

Notes. Spectral (Sa(T), where T is the period in seconds) and peak ground acceleration (PGA) values are given in units of g (9.81 m/s^2). Peak ground velocity is given in m/s. Values are for "firm ground" (NBCC 2015 Site Class C, average shear wave velocity 450 m/s). NBCC2015 and CSAS6-14 values are specified in **bold** font. Three additional periods are provided - their use is discussed in the NBCC2015 Commentary. Only 2 significant figures are to be used. ***These values have been interpolated from a 10-km-spaced grid of points. Depending on the gradient of the nearby points, values at this location calculated directly from the hazard program may vary. More than 95 percent of interpolated values are within 2 percent of the directly calculated values.***

Ground motions for other probabilities:

Probability of exceedance per annum	0.010	0.0021	0.001
Probability of exceedance in 50 years	40%	10%	5%
Sa(0.05)	0.025	0.060	0.091
Sa(0.1)	0.034	0.086	0.134
Sa(0.2)	0.052	0.114	0.166
Sa(0.3)	0.060	0.118	0.163
Sa(0.5)	0.058	0.110	0.146
Sa(1.0)	0.047	0.085	0.109
Sa(2.0)	0.028	0.049	0.063
Sa(5.0)	0.0098	0.018	0.023
Sa(10.0)	0.0039	0.0068	0.0089
PGA	0.022	0.050	0.075
PGV	0.046	0.087	0.114

Figure 2-3: 2015 National Building Code Seismic Hazard Calculation for the Minto Mine

3 METEOROLOGY

The existing meteorology of the Project area and the climate trends and projections of the Minto region of Yukon were first described in detail in EBA's report *Minto Mine – 2010 Climate Baseline Report* and subsequently in Access Consulting Group's (ACG) *Minto Climate Baseline Report* (2012) and Alexco Environmental Group's (AEG) *Minto Mine Site Characterization – Hydrometeorology* (2017) (Appendix 1-2). The meteorological data contained in the above reports were recorded by two on-site meteorological stations over a period of about twelve years. Snow depth and water equivalent data were provided from annual on-site snow surveys since 1994. Data used in climate trend analyses were sourced from regional monthly and annual records provided by the Meteorological Service of Canada and applied to the Project area. Future temperature and precipitation scenarios for the Minto Mine areas were obtained from a probabilistic analysis conducted by SRK (2016). Details of the data collection program and the procedures used for analysis are provided in the 2017 *Hydrometeorology Site Characterization Report* (Appendix 1-2).

The climate in the Minto region is subarctic continental characterized by long, cold winters and short, cool summers. The area experiences moderate precipitation in the form of rain and snow and a large range of temperatures on a yearly basis with a mean annual temperature below 0°C.

3.1 SUMMARY OF SITE METEOROLOGICAL CONDITIONS (2005-2016)

Two meteorological stations installed at the property have recorded wind speed and direction, air temperature, relative humidity, barometric pressure, solar radiation and rainfall in one-hour intervals since September 7, 2005 for the HOBO and October 15, 2010 for the Campbell Scientific station. The Campbell Scientific weather station has been shown to provide more reliable data and as such, the HOBO station was decommissioned in the summer of 2014.

3.1.1 Wind

The HOBO anemometer was installed at a height of 3 meters and got damaged by strong winds in 2010. Based on data from the Campbell Scientific anemometer, which is installed at a height of 10 meters and is much less prone to icing than the HOBO anemometer, dominant winds at site originate from the South, Southeast and Northwest, with an average speed of 2.8 m/s and gusts up to 26.8 m/s.

3.1.2 Air Temperature

Summer is characterized by temperatures in the range of 10°C to 20°C. Winter is characterized by a much larger day-to-day variation in temperatures, typically between –10°C and –30°C. Diurnal variation in air temperatures tends to be less during the winter period than during the summer period. The transitions between winter and summer are characterized by a quick rise or fall in air temperatures. July has been the warmest month on average (14.5°C) while the coldest has been December (–17.7°C). The mean annual

temperature for the period of record is -1.9°C for the HOBO station and -0.9°C for the Campbell Scientific station. Extremes ranged from -43.2°C to 30.3°C.

3.1.3 Relative Humidity

Relative humidity (RH) is highest during the winter months (typically in the range of 75% to 95%) and lowest during the spring and early summer, typically in the range of 40% to 60%. Relative humidity has a much larger day-to-day variability during the summer. The lowest recorded %RH was 10.25%. The highest was 100%. Annually, mean relative humidity is 70%.

3.1.4 Solar Radiation

As would be expected at a latitude of 62.6°N, a strong seasonal pattern is evident in solar radiation, with a maximum being received near the summer solstice in late June (daily maximums on the order of 750 W/m^2), and daily maximums slightly above zero around the winter solstice when the site experiences only about 3 hours of direct sunlight per day. Large fluctuations from the general trend during the summer are due to cloud cover. Mean annual solar radiation is 108 W/m^2 (averaged over the two meteorological stations).

On July 20, 2016, a second pyranometer was installed at the Campbell Scientific station to measure outgoing radiation. Data from August 2016 to December 2016 range from 5 w/m^2 to 34 W/m^2.

3.1.1 Barometric Pressure

Barometric pressure recorded on-site at 885 m elevation is converted into a meteorological standard sea level equivalent. Barometric pressure is slightly higher on average between May and August (above 1009 hPa), and lowest between September and April (below 1008 hPa). Mean annual sea-level equivalent barometric pressure is 1006.5 hPa (average of both meteorological stations).

3.1.2 Precipitation

3.1.2.1 Rainfall

The Minto Mine HOBO meteorological station has been recording rainfall on site since its installation in September 2005 until decommissioning in June 2014. Based on a cumulative of average monthly rainfall, 192.8 mm of rain is expected on average at site in a single year (AEG, 2017). August is the rainiest month with an average rainfall of 46.2 mm. The largest monthly rainfall total was 100.6 mm (August 2008). Rainfall has been recorded in every month of the year. Based on regional data, about 64% of total annual precipitation falls as rain (average proportion over the period of record for four regional stations). Applying

this ratio to the average annual rainfall of 192.8 mm measured at the Minto Mine HOBO station would result in a total annual precipitation value of 303.2 mm.

3.1.2.2 Total Precipitation

The Campbell Scientific station was recording rainfall only from time of commissioning until October 14, 2011, when a snowfall conversion adaptor was installed on the tipping bucket to allow for measurement of total precipitation. On November 18, 2014, a Geonor T-200B vibrating wire all weather precipitation gauge was installed in addition to the existing tipping bucket and rainfall adaptor to improve accuracy and validate total precipitation measurements. Average annual total precipitation is 277.6 mm with the tipping bucket and snowfall adaptor and 248.5 mm with the total precipitation gauge. Both values are less than the estimated total of 303.2 mm based on the HOBO rainfall measurements and regional data. This could indicate potential undercatch of both the snowfall adaptor and total precipitation gauge, however, other factors such as wind or snow pillow formation can also play a role. More years of total precipitation data will allow a more meaningful comparison.

3.1.3 Snowpack

Annual snow surveys have been conducted in 1994, 1995, 1998, and annually from 2006 to 2016, at three locations in the Minto Creek catchment area and a fourth station was added in 2016. Based on the ten years of snow surveys, the average water-equivalent snow depth remaining on the first day of March and April is 86.3 mm and 86.7 mm, respectively. In four of the seven May surveys, the snowpack had melted entirely by May 1. In the three years where snow remained on the ground on May 1 (1995, 2006, and 2008), the mean snowpack had reduced to 10%, 27%, and 20%, respectively, of the peak measured snowpack in that year. The data indicates that the majority of runoff due to snowmelt occurs in April.

3.1.4 Evaporation

No evaporation records are available for the Minto Mine site. An evaporation pan was deployed on site during the summer of 2016, however, due to a system error, data are invalid. Starting in August 2012, the Campbell Scientific datalogger program incorporated an instruction for the calculation of potential evapotranspiration (PET) using the American Society of Civil Engineers (ASCE) standardized reference evapotranspiration equation (Penman-Monteith). The PET instruction uses temperature, relative humidity, wind speed, solar radiation, latitude, longitude and altitude to calculate an evaporation rate for a short grass crop. This provides an approximation of actual evaporation, which varies locally depending on surface type and micro topography. The average total annual PET is 428 mm which is similar to the estimated mean evaporation value of 400 mm/year based on regional data.

3.2 REGIONAL CLIMATE TRENDS

Long-term trends were evaluated by reviewing long-term records from stations in the region (regional analysis). Environment Canada meteorological stations used for this regional analysis were selected based on location/proximity and data availability.

3.2.1 Temperature

3.2.1.1 Past Trends

The annual mean temperature has been increasing at Pelly Ranch by 0.07°C per year on average over the 57-year period corresponding to a total average increase of 4.2°C from 1957 to 2014. For Carmacks, the average rate of increase of the annual mean temperature is slightly less at 0.05°C per year, resulting in an average increase of 2.5°C for the 52-year period. It is also important to note that the average annual minimum temperature has been increasing at a faster rate than the average annual maximum temperature at both stations, thus implying that the average diurnal range has also been decreasing. With respect to seasonality, January (winter) mean temperatures have experienced much higher rates of increase (0.15°C per year at Pelly Ranch and 0.22°C per year at Carmacks) than July (summer) mean temperatures (0.03°C per year at Pelly Ranch and 0.02°C per year at Carmacks). A close correlation between monthly average temperatures recorded at Minto and at Pelly ranch and Carmacks, as well as the proximity of the stations to the property, suggests that the long term trends would also be applicable to the Minto property. Winter temperatures at Minto however are approximately 3°C to 5°C higher due to a predominant Yukon winter temperature inversion of +8°C/ km up to an elevation of 1200 m (EBA, 2010).

3.2.1.2 Future Trends

Based on a probabilistic analysis conducted by SRK (2016), the mean annual temperature at Minto is expected to increase by about 3.3°C over the next century, with mean annual temperatures of -0.8°C, 0.2°C and 1.1°C in the 2020s, 2050s and 2080s, respectively. The intra-annual range between extreme maximum and extreme minimum temperatures is expected to decrease with time (SRK, 2016).

3.2.2 Precipitation

3.2.2.1 Past Trends

Environment Canada monthly precipitation records at Pelly Ranch and Carmacks show an increasing trend in total annual precipitation over the period of record, however the correlation is weak. The proportion of total

precipitation falling as rain also displays an increasing trend over the period of record, but again, the correlation is weak.

3.2.2.2 Future Trends

Based on a probabilistic analysis conducted by SRK (2016), The total precipitation at Minto is forecasted to increase 67.9 mm (15%) during the mine life (2011 to 2040) and 199.3 mm (44%) by 2100 the total precipitation at Minto is forecasted to increase 67.9 mm (15%) during the mine life (2011 to 2040) and 199.3 mm (44%) by 2100. The simple daily precipitation intensity index (mean annual precipitation divided by the total number of wet days) is forecasted to increase by 6.6% during the mine life and by 25% in the next century. Snow accumulation appears to be increasing between December and March mainly by years 2100s.

4 SURFACE WATER HYDROLOGY

The baseline surface water hydrology of the Minto Creek watershed prior to mining activity was detailed in Clearwater Consultants Ltd. Memorandum CCL-MC6 *Minto Copper Project—Surface Water Hydrology Conditions* (CCL, 2010). CCL-MC6 also covered conditions during Mine Operations until 2009. A subsequent report was prepared by ACG to update CCL-MC6 by presenting those previous data, and data gathered since then until the end of the 2012 open water season (ACG, 2013). Additionally, this report included data since 2009 gathered in McGinty Creek, a catchment of similar size to Minto Creek, also draining into the Yukon River and located directly north of the Minto Creek catchment (Figure 4-1). An updated *Minto Mine Site Characterization – Hydrometeorology* report prepared by Alexco Environmental Group (2017) is presented in Appendix 1-2.

Hydrological data have been gathered by either ACG or Minto representatives. Data coverage from year to year varies, depending on when in situ dataloggers were installed and removed, and when instantaneous discharge measurements were taken. Instantaneous discharge is measured using the velocity-area method and a current meter. Solinst Water Leveloggers are used to collect continuous stage readings which are then corrected based on physical staff gauge measurements. The records are processed into continuous discharge based on the stage-discharge relationship. This relationship (stage and discharge) is established each season through rating measurements obtained during regular field visits to the sites.

The locations for which the greatest amount of data have been collected are stations W1, Minto Creek near the mouth (catchment area of 42 km^2); and W3, Minto Creek downstream of water storage pond dam (catchment area of 10.4 km^2 area). In 2010, another continuously monitored hydrometric site called MC1 was added, approximately 2 km upstream of W1. In 2011, data collection did not allow for processing of stage records into continuous discharge; however, improved monitoring allowed for successful processing in 2012.

In addition to Minto Creek, the catchment to the north has been monitored since 2009 to support development of the Minto North deposit, and is referred to as McGinty Creek. There are five stations on McGinty Creek at which discharge is measured. In 2009, four stations were established including MN-4.5, MN-2.5, MN-1.5 and MN-0.5; in 2011 a fifth site, MN-0.2, was added (Figure 4-1). Continuous hydrometric data has been collected at MN-4.5 since 2009, and additional continuous logging instruments were added to stations MN-2.5 and MN-0.5 in late 2012.

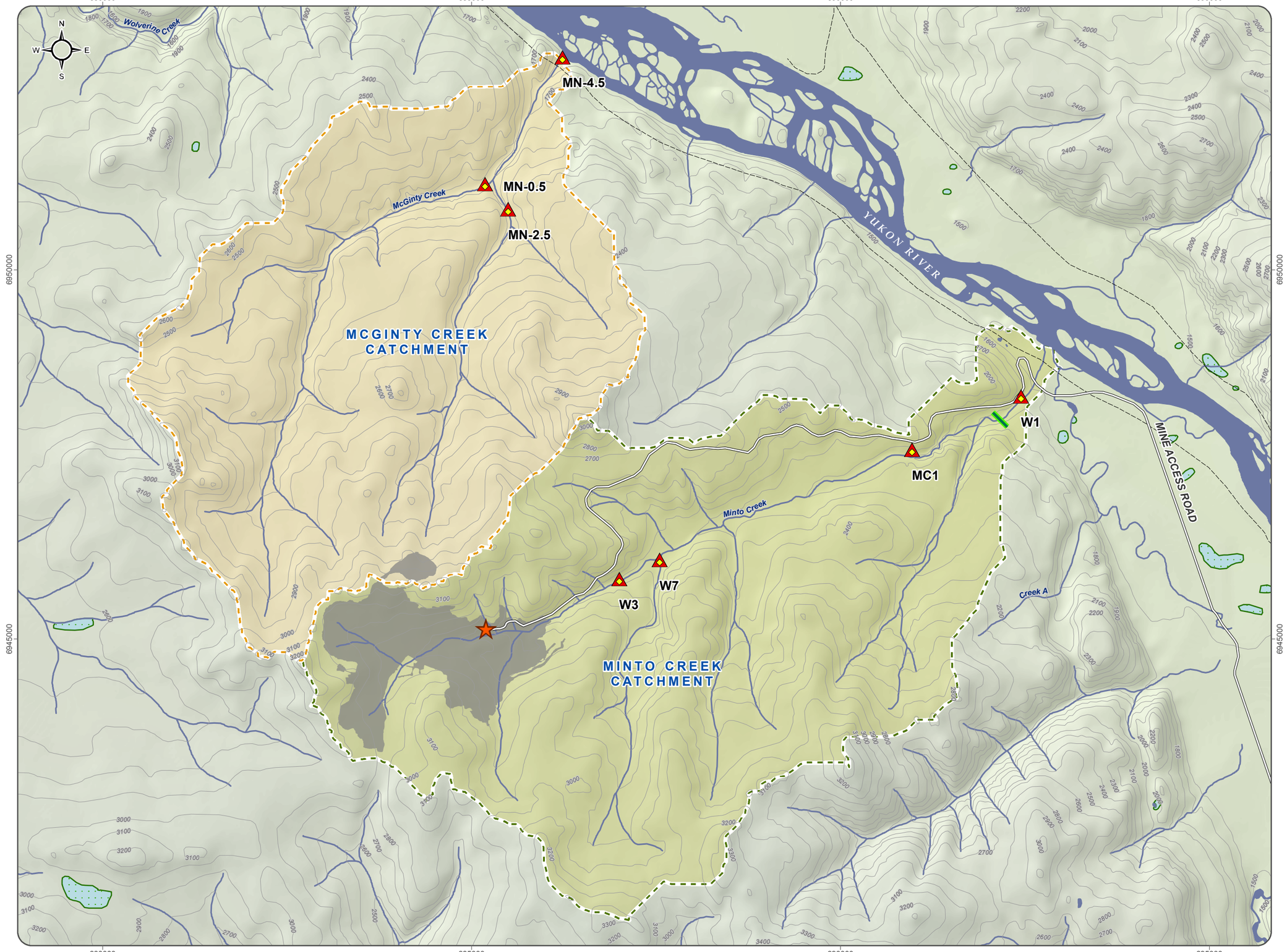
MINTO MINE

CAPSTONE MINTO MINE - SITE CHARACTERIZATION PLAN

FIGURE 4-1

MINTO AND MCGINTY CREEK HYDROMETRIC MONITORING NETWORK AND CATCHMENT AREAS

JANUARY 2018



- Water Quality and Hydrometric Monitoring Station
- Observed Fish Barrier - Minto Creek
- Mine Footprint
- McGinty Creek Catchment
- Minto Creek Catchment
- Wetland
- Minto Access Road
- Contours (100ft)

1:50,000 *when printed on 11 x 17 inch paper*

0 500 1,000 1,500 2,000 Meters

Alexco Environmental Group

National Topographic Data Base (NTDB) compiled by Natural Resources Canada at a scale of 1:50,000. Cadastral data compiled by Natural Resources Canada. Reproduced underlicense from Her Majesty the Queen in Right of Canada, Department of Natural Resources Canada. All rights reserved.

Datum: NAD 83 Projection: UTM Zone 8N

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(*Last edited by: amatlashevska;12/01/2018/11:24 AM*)

4.1 MINTO CREEK

Monitoring of hydrological parameters on Minto Creek began in 1993 and has continued intermittently at sites W1 and W3 (Figure 4-1). Monitoring has been more intensive since mine commissioning in 2007. W3 is an in-stream trapezoidal flume with a manufacturer-specified stage-discharge relationship. Both discharge and stage are read on an integrated gauge in the throat of the flume. A Solinst Levelogger record is calibrated with these field observations to process a continuous discharge record at this site. Flows at W3 are impacted by storage within, and discharge from the Water Storage Pond. Sites MC-1 and W1 are natural stream channels where manual velocity measurements are taken across the channel and discharge is calculated using the velocity area method. Continuous water levels from Solinst Leveloggers are processed into continuous discharge using these rating measurements. Updated mean monthly flows for W1 and W3 are presented in Table 4-1 and Table 4-2, respectively.

Table 4-1: Mean Monthly Discharge (m^3/s) on Minto Creek at Station W1.

	Apr	May	Jun	Jul	Aug	Sep	Oct
1993						0.069	
1994		0.312	0.058	0.095	0.007	0.073	
1995		0.027	0.001	0.091		0.133	
1996		0.031	0.024	0.324		0.146	
1997		1.447			0.265		
1998		0.161			0.003		
1999					0.033		
2000		1.004					
2001		0.467					
2002							
2003					0.129		
2004				0.118			
2005		0.097	0.012	0.127	0.209	0.219	0.134
2006	0.203	0.354	0.15	0.02	0.0068		0.031
2007	0.645	0.175	0.053	0.061	0.025	0.034	0.035
2008		0.117	0.015	0.026	0.184	0.184	0.026
2009		0.868	0.351	0.249	0.139	0.026	
2010	0.560	0.081	0.038	0.106	0.118	0.125	0.092
2011			0.229	0.200	0.200	0.082	
2012		0.269	0.073	0.052	0.051	0.078	0.056 [2]
2013		0.485 [1]	0.064	0.065	0.044	0.085	0.059 [2]
2014		0.138 [1]	0.022	0.020	0.014	0.031	0.025 [2]
2015		0.117	0.010	0.010	0.030	0.024	0.020 [1]
2016	0.252 [1]	0.047	0.020	0.017	0.024	0.043	0.026
Mean							
Pre-Mine 1993 to 2006	0.203	0.433	0.049	0.129	0.093	0.128	0.083
Mining Period 2007 to 2016	0.486	0.255	0.088	0.081	0.083	0.071	0.042
All Data 1993 to 2016	0.415	0.344	0.075	0.099	0.087	0.090	0.050

Notes: [1] Based on incomplete or derived data

[2] Based on multiple discrete measurements

Table 4-2: Mean Monthly[1] Discharge (m^3/s) on Minto Creek at Station W3.

	Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec
1993									0.028			
1994					0.101	0.028	0.039	0.011	0.028			
1995						0.004	0.017		0.027	0.008		
1996					0.013		0.087		0.021			
1997					0.554							
1998								0.006				
1999								0.006				
2000												
2001					0.160							
2002												
2003								0.037				
2004							0.026					
2005					0.046	0.008	0.014	0.017	0.022	0.020		
2006				0.018	0.128	0.042	0.006	0.015	0.009	0.010		
2007				0.001	0.012	0.009	0.006					
2008								0.064[2]	0.122[2]	0.003		
2009						0.026[2]	0.106[2]	0.092[2]	0.124[2]	0.110		
2010				0.002	0.004	0.005	0.034	0.071	0.086	0.070		
2011						0.005	0.005	0.006	0.005			
2012				0.004	0.020	0.003	0.004	0.004	0.004	0.004		
2013						<0.001	<0.001	0.002	0.003	0.003	0.003	0.006
2014	0.003	0.006	0.006	0.057	0.086	0.003	0.003	0.003	0.004	0.004	0.004	0.006
2015	0.004	0.002	0.001	0.075	0.052	0.003	0.003	0.004	0.004	0.004	0.004	0.003
2016				0.022	0.052	0.003	0.003	0.004	0.004	0.004	0.003	0.004
Mean												
Pre-Mine 1993 to 2006	-[3]	-[3]	-[3]	-[3]	0.167	0.02	0.032	0.015	0.023	0.013	-[3]	-[3]
Mining Period 2007 to 2016	0.004	0.004	0.004	0.027	0.038	0.007	0.021	0.028	0.040	0.025	0.004	0.005
All Data 1993 to 2016	0.004	0.004	0.004	0.026	0.102	0.012	0.025	0.023	0.033	0.022	0.004	0.005

[1]Monthly flows calculated by averaging all available flow data for a given month. Average flow in months with only a single spot flow measurement assumed equal to the spot flow measurement.

[2]Flows impacted by storage within, and emergency releases from, the Water Storage Pond in August and September 2008 and in June through October 2009.

[3]Insufficient data for calculation.

Table 4-3 shows the discharge data gathered to date at station MC-1 and Table 4-4 shows data available at station W7. Appendix 1-2 contains the associated baseline report and the available hydrographs from 2007-2012 for stations W1, W3, W7 and MC-1.

Table 4-3: Mean monthly discharge (m^3/s), Minto Creek at MC-1.

	Month					
Year	**May**	**Jun**	**Jul**	**Aug**	**Sep**	**Oct**
2011					0.118	0.093
2012	0.179	0.065	0.052	0.041	0.108	-
2013	0.358	0.085	0.103	0.044	0.089	0.064 [2]
2014	0.187	0.028	0.031	0.036	0.028	0.033 [2]
2015	0.862 [2]	0.014	0.028	0.042	0.035	0.031 [1]
2016	-	0.029 [1]	0.036	0.048	0.076	-
Mean	**0.397**	**0.044**	**0.050**	**0.042**	**0.076**	**0.055**

Notes: [1] Based on incomplete or derived data
[2] Based on multiple discrete measurements

Table 4-4: Mean monthly discharge (m^3/s), Minto Creek at W7.

	Month					
Year	**May**	**Jun**	**Jul**	**Aug**	**Sep**	**Oct**
2013			0.013	0.031	0.019	0.006
2014	0.112					
2015		0.006	0.004	0.011	0.011	0.029
2016		0.006	0.010	0.010	0.013	0.010

Notes: Based on discrete measurements only

4.2 McGinty Creek

McGinty Creek has two main sub-catchments that each have two water quality monitoring stations, one just above the confluence and one near the headwaters. MN-4.5 is located on the main stem below the confluence of the tributaries near the mouth; just above the Yukon River (Figure 4-1). MN-0.5 and MN-0.2 are the lower and upper stations on the west tributary, respectively. MN-2.5 and MN-1.5 are the lower and upper stations on the east tributary, respectively. Stations MN-0.2 and MN-1.5 are not set up with continuous loggers and generally exhibit very low flows. As such they are not included in the discussion below.

Table 4-5: Mean monthly discharge (m^3/s), McGinty Creek at MN-4.5.

	Month						
Year	**Apr**	**May**	**Jun**	**Jul**	**Aug**	**Sep**	**Oct**
2009	-	0.018 [1]	0.033	0.019	0.031	0.016	0.013 [1]
2010	-	0.028 [1]	0.051	0.079	0.047	0.034	-
2011	-	0.444	0.093	0.125	0.134	0.068	0.045 [1]

	Month						
Year	**Apr**	**May**	**Jun**	**Jul**	**Aug**	**Sep**	**Oct**
2012	0.212 [1]	0.230	0.180	0.082	0.053	0.109	-
2013	-	-	0.054 [1]	0.103	0.093	0.116	-
2014	-	0.230 [1]	0.041	0.037	0.026	0.046 [1]	-
2015	-	-	0.013	0.046	0.049	0.029	0.029 [1]
2016	-	0.017 [1]	0.015	0.026	0.028	0.028	0.010 [1]

Notes: [1] Based on incomplete or derived data

Table 4-6: Mean monthly discharge (m^3/s), McGinty Creek at MN-2.5.

	Month						
Year	**Apr**	**May**	**Jun**	**Jul**	**Aug**	**Sep**	**Oct**
2014 [1]	-	0.032	-	-	-	-	-
2015 [1]	-	0.015	0.003	0.011	0.038	0.010	0.010
2016 [1]	0.034	0.008	-	-	0.009	0.014	0.003

Notes: [1] Based on incomplete or derived data

Table 4-7: Mean monthly discharge (m^3/s), McGinty Creek at MN-0.5.

	Month						
Year	**Apr**	**May**	**Jun**	**Jul**	**Aug**	**Sep**	**Oct**
2014 [1]	-	0.045	-	-	-	-	-
2015 [1]	-	0.035	0.013	0.054	0.098	0.029	0.025
2016 [1]	-	0.028	0.016	0.010	0.016	0.035	0.018

Notes: [1] Based on incomplete or derived data

Hydrographs from 2009–2016 at MN-4.5, MN-2.5 and MN-0.5 are shown in the *Minto Mine Site Characterization – Hydrometeorology* report (Appendix 1-2).

5 SURFACE WATER QUALITY

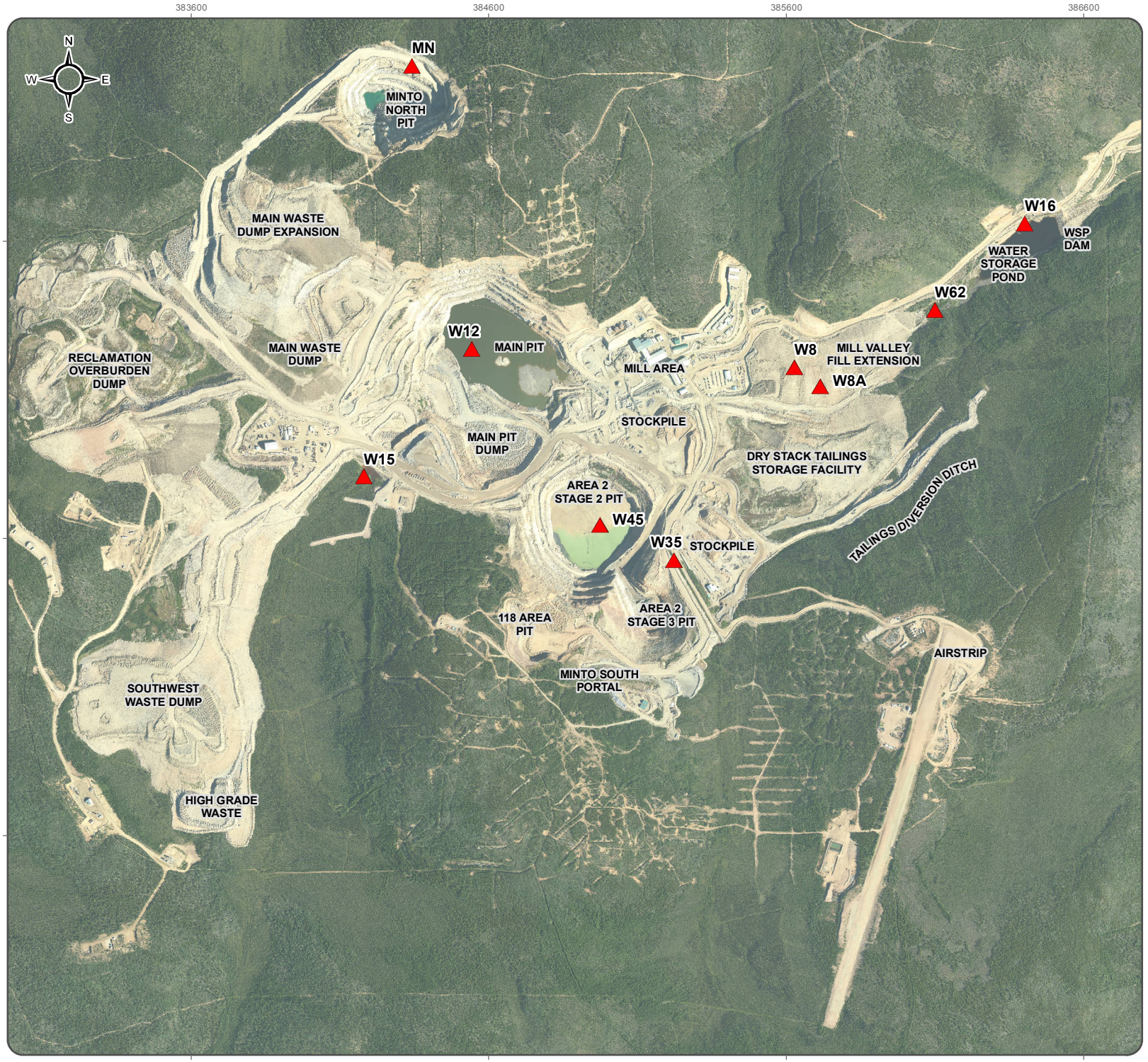
Surface water quality is a key consideration in the evaluation of potential effects of mining and mineral development projects. Effects from mining activities can be observed for significant distances downstream and changes to water quality parameters have the potential to impact aquatic resources and to affect human use of water resources. Both Minto and McGinty Creeks are receiving environments for Minto Mine and have been monitored for water quality conditions, with more intensity and duration of sampling in Minto Creek as part of the initial baseline and more recently the intensive operational monitoring under the mine water quality monitoring programs (licenced and otherwise). A characterization report for water quality in these two watersheds has been developed with the intention of presenting either baseline or existing water quality conditions as the foundation of assessing potential effects from proposed mine expansion activities. Minto Creek and McGinty Creek have both been previously characterized in numerous reports. The *Minto and McGinty Creeks Water Quality Characterization report* (AEG, 2017) included as Appendix 1-3 provides updates to the water quality characterization of both Minto and McGinty Creek stations and for key mine site water management and monitoring stations, and includes monitoring results to December 2016. Key findings from the characterization report are included below.

5.1 MINTO MINE SITE

The water quality for ten monitoring stations around the Minto Mine Site including W8, W8A, W12, W15, W16, W35, W35A, W35B, W45 and W62 is summarized in this section. For ease of discussion and presentation of results, these stations are divided into two groupings: the “upper mine site” and “lower mine site”. The Minto Mine Site stations and groupings are described in Table 5-1 and shown in Figure 5-1. Water quality is monitored at many more locations on the Minto site, but these ten stations are the most critical with respect to decisions regarding the conveyance, storage and treatment of site water.

Table 5-1: Minto Mine Monitoring Station Locations.

Station	Description / Location
Upper Mine Site	
W12	Main Pit
W15	Minto Creek, d/s of Southwest Waste Rock Dump
W35	South Diversion Ditch
W35A	North End of South Diversion Ditch
W35B	South End of South Diversion Ditch
W45	Area 2 Pit
Lower Mine Site	
W8	Dry Stack Tailings Storage Facility drainage, west
W8A	Dry Stack Tailings Storage Facility drainage, east
W16	Water Storage Pond
W62	Sump at toe of Mill Valley Fill Extension



MINTO MINE

CAPSTONE MINTO MINE - SITE CHARACTERIZATION PLAN

FIGURE 5-1

MINE SITE MONITORING STATION LOCATIONS

JANUARY 2018

capstone MINTO MINE

▲ Surface Water Surveillance Program Monitoring Station Locations

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Aerial imagery obtained from Challenger Geomatics. Imagery acquired August 3rd th 2017.

Datum: NAD 83 Projection: UTM Zone 8N

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The characteristics, trends and patterns associated with the geochemistry of surface waters at the Minto Mine Site are summarized below, while more details are available in *Minto and McGinty Creeks Water Quality Characterization report* (AEG, 2017) included as Appendix 1-3.

- The pH of the waters was predominantly circumneutral to slightly alkaline;
- Although typically well oxygenated, lower levels of dissolved oxygen were often observed during winter, particularly for the pit lake locations, where ice cover limited the oxygenation of underlying waters;
- Nitrogen species concentrations were generally highest in the pit lake and DSTSF drainage sites, reflecting leaching of nitrogen residue from blasting activity. Nitrate-N levels showed marked seasonality at both the Minto Creek mine site location (W15) and the DSTSF drainage (W8 and W8A), but with different patterns. In Minto Creek, nitrate-N levels were highest in winter and lowest over summer, responding to nitrogen uptake demand by primary producers, whereas the opposite trend was noted in the DSTSF drainage;
- DOC concentrations were highest during spring snowmelt, likely due to snowmelt flushing the surrounding soils of labile organic matter that had accumulated through the previous summer and fall;
- Dissolved hardness concentrations were typically lowest in spring at the flowing water sites due to snowmelt dilution, and increased through the year, peaking in winter, suggestive of a marked groundwater component to hardness levels;
- Concentrations of dissolved iron, and to a lesser extent dissolved aluminum, often appeared higher than may be expected from equilibrium with their oxyhydroxide minerals under the prevalent oxidizing, circumneutral pH conditions. Complexation with dissolved organic matter likely maintains these elevated dissolved iron and aluminum levels;
- Dissolved copper concentrations were highest in the DSTSF drainage and tended to peak with freshet, suggestive of some association with DOC; and
- Dissolved concentrations of both selenium and molybdenum increased approximately four-fold in the pit lake sites over the past two to three years, alongside sulphate. Dissolved selenium exhibited strong seasonal cycling in both Minto Creek (W15) and the DSTSF drainage (W8A), but with offset patterns. In Minto Creek, selenium concentrations were lowest at freshet, then rose through the year, peaking in winter, suggestive of a strong groundwater contribution. The opposite pattern was observed at site W8A, where selenium levels oscillated on an annual basis between summer peaks and winter troughs.

5.2 MINTO NORTH PIT

The Minto North Pit is located at the northernmost tip of the mine site (Figure 5-1). Station MN was established with the initiation of the mining activities in the Minto North Pit (August 2015), and the monitoring results presented in the report for this station are all 'operational' results. Water quality data from the Minto North Pit are considered in isolation since the pit does not discharge at present and is not anticipated to discharge during the operational period.

The characteristics, trends and patterns associated with the geochemistry of Minto North Pit are summarized below, while more details are available in *Minto and McGinty Creeks Water Quality Characterization report* (AEG, 2017) included as Appendix 1-3.

- Minto North Pit waters were circumneutral to mildly alkaline and well oxygenated over the one year of data collected to date;
- Field-measured conductivity and hardness were relatively high and showed no seasonal trends;
- Nitrogen species were very high in the Minto North Pit, likely due to nitrogen residue from blasting activities;
- Minto North Pit waters generally contained elevated dissolved copper, molybdenum, and selenium concentrations;
- The majority of dissolved silver and zinc measurements were below detection; and
- Dissolved aluminum, arsenic, cadmium, nickel, and iron concentrations were unremarkable although some sporadic spikes in iron levels were an order of magnitude higher than the median dissolved iron concentration.

5.3 MINTO CREEK

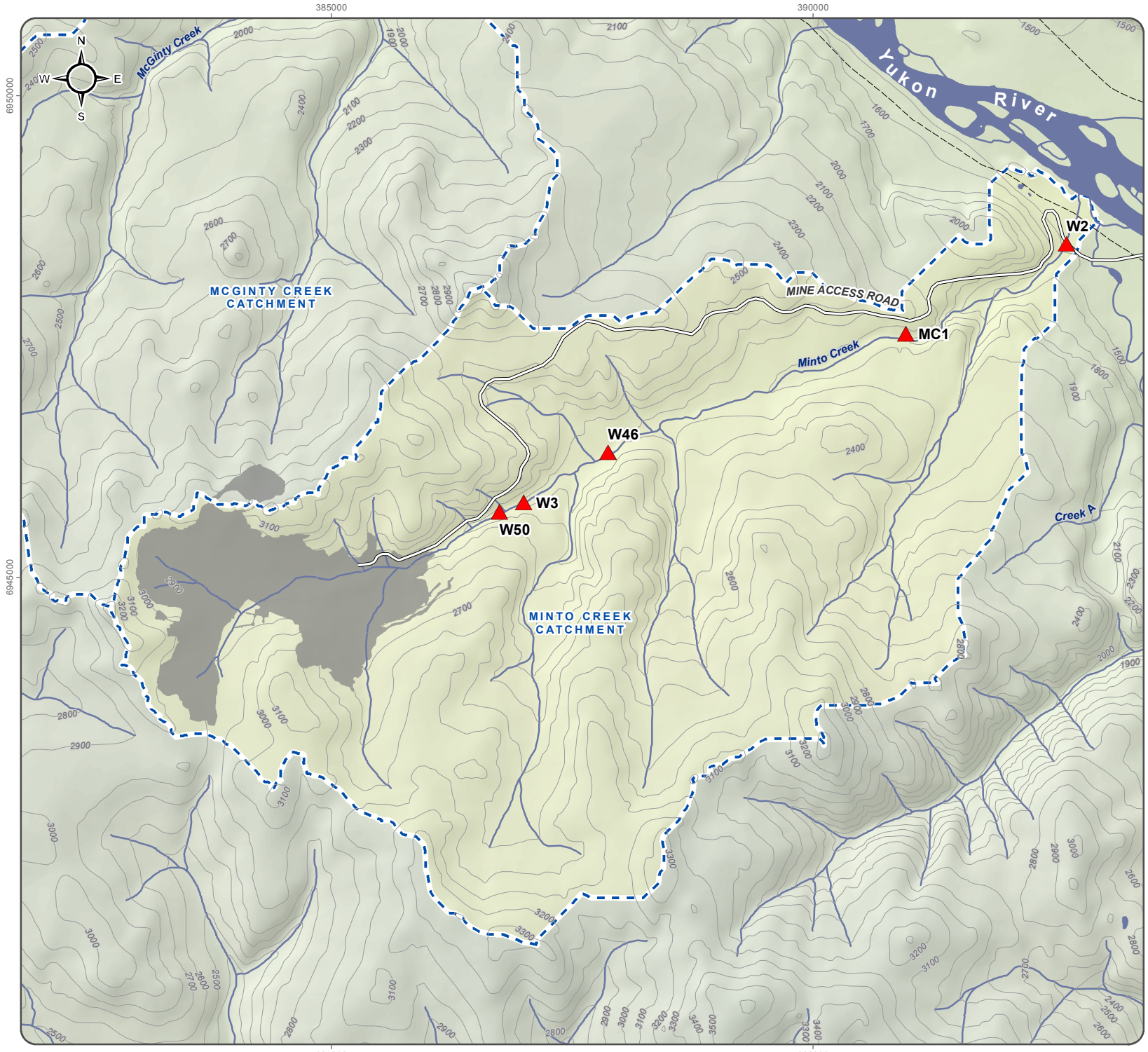
Several water quality stations are monitored on Minto Creek during the operational period. Data from four of these monitoring stations are provided in this report: W50, W3, MC1, and W2. These stations are described in Table 5-2 and their location shown in Figure 5-2. These stations are all on the 'mainstem' of Minto Creek.

Table 5-2: Minto Creek Monitoring Station Locations

Station	Description / Location
W50	Minto Creek, approximately 50m downstream of toe of WSP dam
W3	Downstream of Water Storage Pond dam, MMER Final Discharge Point (Flume) – Effluent
MC1	Receiving Environment station upstream of the canyon fish barrier on Minto Creek.
W2	Lower Minto-Creek at Road Crossing – Receiving Environment station in reach with documented fisheries usage.

W50 is located downstream of the inflow of the effluent discharge and is a point of compliance for effluent quality standards (EQSs) as outlined in WUL QZ14-031 (August 5th, 2015). EQSs are presented and discussed in further detail in Section 3.3 of Appendix 1-3 *Minto and McGinty Creeks Water Quality Characterization report* (AEG, 2017). Station W3—located downstream of the Water Storage Pond dam—is the mine effluent monitoring and compliance point per the Metal Mining Effluent Regulations (MMER). Station W2 is the lowermost Minto Creek receiving environment monitoring location before the confluence with the Yukon River, and is the "WQO Station" designated in the Water Use Licence QZ94-031 during the period when flow is encountered at stations W15 and W35. Both W2 and W3 have been monitored regularly since 2005. Station MC1 in Minto Canyon serves as a comparison monitoring point between W3 and W2 and began regular monitoring in 2010.

Four additional stations (W6, W7, C4 and C10) situated on tributaries to Minto Creek are monitored regularly or semi-regularly. The data for these stations are not included in this reporting update as they were a component of the Minto Creek background water quality evaluation in 2016, and are included in Minnow Environmental Inc. (2016) (Appendix A of Appendix 1-3).



MINTO MINE

CAPSTONE MINTO MINE - SITE CHARACTERIZATION PLAN

FIGURE 5-2

MINTO CREEK MONITORING STATION LOCATIONS

JANUARY 2018

capstone MINTO MINE

- Surface Water Surveillance Program Monitoring Station Locations
- Catchment
- Mine Footprint
- Minto Access Road
- Trail

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5.3.1 Background Water Quality

In 2009, to support the redevelopment of the mine Water Management Plan, Minnow Environmental compiled Minto Creek water quality monitoring data into a pooled data set that reflected background (unimpacted by development activities) conditions. The mine water-quality monitoring program has continued to monitor stations within the operational area, downstream stations which are exposed to the influence of mine effluent discharge, and downstream stations which are not exposed to effluent. Data from these non-exposed stations are relevant the background water quality condition in Minto Creek, and were therefore added to the pre-existing 2009 background data set. New background data sets were developed for periods up to the end of 2010 and to the end of 2012. The background water quality data are also used in the water quality prediction work as the 'unimpacted' source term for undisturbed catchment areas.

5.3.2 Mainstem Minto Creek Water Quality

The characteristics, trends and patterns associated with the geochemistry of surface waters at the Minto Creek stations is summarized below, while more details are available in *Minto and McGinty Creeks Water Quality Characterization report* (AEG, 2017) included as Appendix 1-3.

- The Minto Creek waters were typically circumneutral to slightly alkaline pH and well-oxygenated;
- Nitrogen species concentrations were generally higher during and following periods of mine discharge, likely due to blasting activity within the mine site. Aside from discharge periods, seasonal trends were observed, particularly for nitrate-N, with lower nitrate-N concentrations typically observed in summer months when primary producers that uptake nitrogen species would be most active;
- Concentrations of DOC at W3 were typically higher during April/May likely due to high flows caused by spring meltwater mobilizing organic carbon that had accumulated over the previous year. DOC concentrations also appeared to be higher during mine discharge periods, but as many of these periods occurred in early spring it is difficult to discern the effects of mine discharge on DOC. At downstream stations MC1 and W2, DOC concentrations were higher in mid to late summer and were generally lower in spring and winter;
- Hardness was often lowest in April and May and increased through the summer and fall suggesting dilution in spring, followed by a higher proportional contribution of groundwater to Minto Creek flow through summer, fall, and winter. This trend was more marked at W50 and downstream stations MC1 and W2, whereas W3 returned higher hardness concentrations with less seasonality;

- Dissolved aluminum concentrations were higher in the spring and decreased through the year, likely due to mobilization by high flows from spring meltwater and possibly complexation by DOC, which also peaked in spring;

- Dissolved arsenic and iron concentrations were typically higher at downstream stations MC1 and W2 than upstream stations and were highest in mid to late summer. A reasonable correlation with DOC concentrations suggests DOC likely enhances iron solubility at MC1 and W2;

- Since 2010, dissolved chromium, lead, silver and zinc has generally been closer to or below detection levels with and without mine discharge;

- Dissolved copper, molybdenum and selenium concentrations were higher during and following periods of mine discharge. Dissolved copper concentrations also appeared to vary seasonally with higher concentrations observed in the spring with higher flows from meltwater and were correlated with DOC concentrations. Dissolved molybdenum concentrations were consistently higher at W3 than W2 in pre-operations and during non-discharge periods.

5.4 McGinty Creek

McGinty Creek water quality was monitored monthly by ACG from May 2009 until July 2012 and has since been monitored monthly by Minto Mine Environment Department staff.

Water quality is monitored at five stations in the McGinty Creek watershed, described in and shown in . Water quality results for McGinty Creek stations are presented and discussed in two groups:

1. East arm of McGinty Creek (MN-1.5 and MN-2.5) and MN-4.5; and

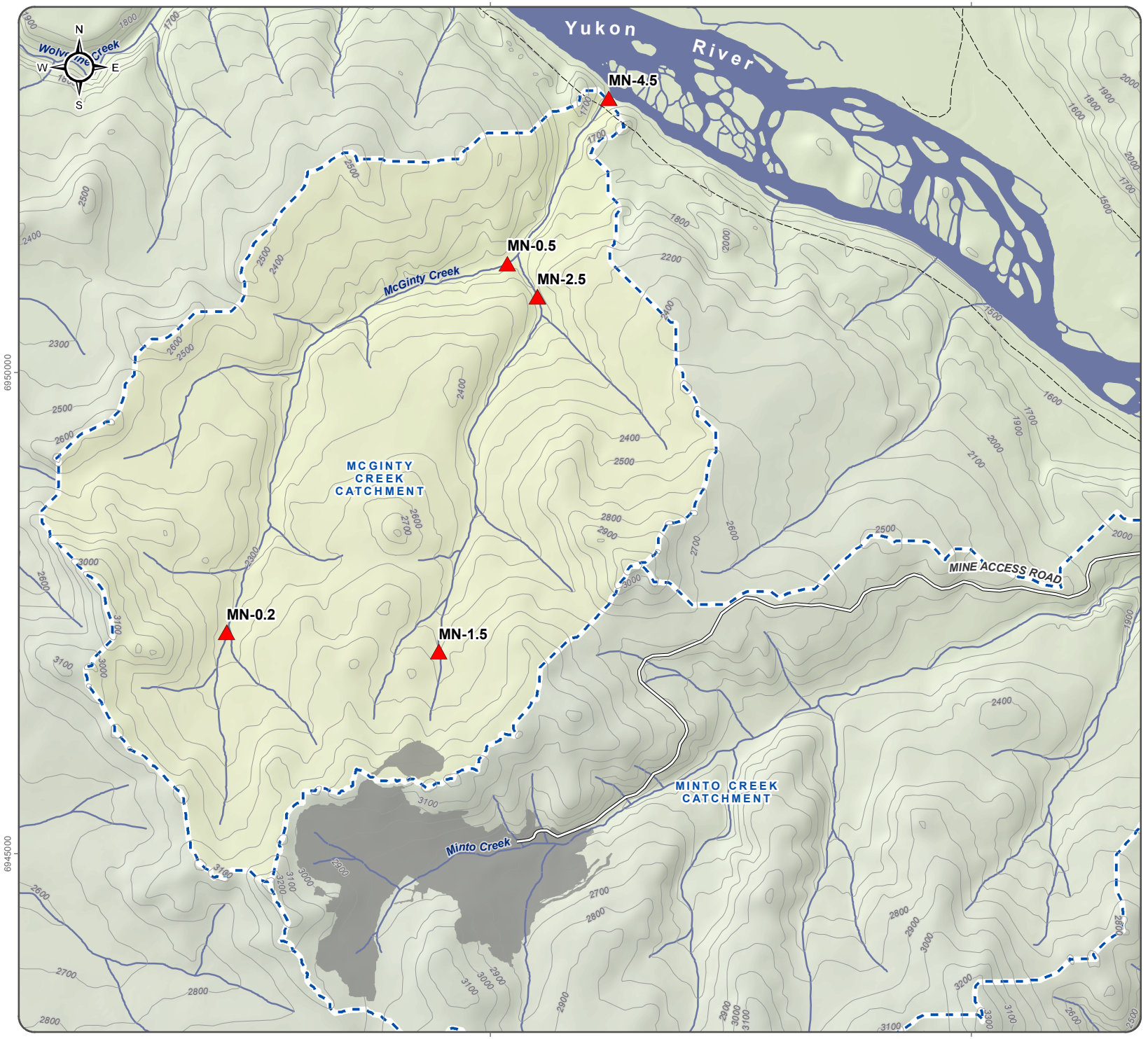
2. West arm of McGinty Creek (MN-0.2 and MN-0.5).

The east arm of McGinty Creek is considered the ‘exposure tributary’ as it originates downgradient of the Minto North deposit where the Minto North pit is located, while the west arm of McGinty Creek is considered the ‘reference tributary’.

Table 5-3: McGinty Creek Monitoring Station Locations.

Station	Description / Location
East Arm of McGinty Creek	
MN-1.5	Upper east arm of McGinty Creek downstream of the Minto North deposit
MN-2.5	East arm of McGinty Creek just upstream of confluence with the west arm
MN-4.5	Lower mainstem McGinty Creek near confluence with Yukon River
West Arm of McGinty Creek (Reference Stations)	
MN-0.2	Upper west arm of McGinty Creek
MN-0.5	West arm of McGinty Creek just upstream of the confluence with the east arm

Monitoring at the upper reference station MN-0.2 was suspended between June 2009 and April 2011, which explains in part the lower total number of samples collected at this station. Similar to Minto Creek, McGinty Creek typically freezes in the winter months with little to no flow observed at all monitoring stations.



MINTO MINE

RECLAMATION AND CLOSURE PLAN

FIGURE 4-4

MCGINTY CREEK WATER QUALITY MONITORING STATION LOCATIONS

JANUARY 2018

capstone MINTO MINE

- Surface Water Surveillance Program Monitoring Station Locations
- Catchment
- Mine Footprint

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The characteristics, trends and patterns associated with the geochemistry of surface waters at the and McGinty Creek stations is summarized below, while more details are available in *Minto and McGinty Creeks Water Quality Characterization report* (AEG, 2017) included as Appendix 1-3.

- Field pH was generally circumneutral to slightly alkaline and increased farther downstream in both arms of McGinty Creek;
- The waters were fairly well-oxygenated with occasional troughs of dissolved oxygen measured in winter months when ice cover likely inhibited ingress of oxygen to McGinty Creek;
- Nitrate-N concentrations were typically lower in April/May due likely to dilution from meltwater and higher values were typically observed in the winter when uptake by primary producers would have been lowest. Nitrite-N was predominantly below detection level in McGinty Creek;
- Concentrations of DOC were typically higher during April/May likely due to mobilization by high flows caused by spring meltwater. Higher concentrations were also commonly observed in mid to late summer;
- Dissolved hardness concentrations increased farther downstream and varied seasonally with lowest concentrations in the spring and increasing concentrations through to the winter, suggesting a significant proportion of hardness from baseflow that is diluted by spring meltwater;
- Dissolved aluminum concentrations were higher in the upstream stations of the east and west arms of McGinty Creek and were erratic, perhaps varying with precipitation events as concentrations were typically higher in April and May when higher flows caused by spring meltwater could mobilize aluminum. Complexation by DOC, which was highest during spring, may also have maintained elevated dissolved aluminum concentrations at this time;
- Dissolved arsenic and iron concentrations were higher in upstream stations of the east and west arm of McGinty Creek and often peaked in August or September. A negative correlation between iron and ORP suggests the presence of a groundwater seep with reducing conditions that could have provided elevated concentrations of iron to McGinty Creek. DOC could also be a factor in maintaining higher iron concentrations in solution;
- Dissolved cadmium and zinc concentrations appeared to decrease from 2010 to 2013 and have been fairly low throughout all McGinty Creek stations post-2013;
- Dissolved chromium concentrations were higher in upstream stations of the east and west arms of McGinty Creek;
- The upstream-most stations – MN-1.5 (east arm) and MN-0.2 (west arm reference tributary) – consistently returned higher dissolved copper concentrations in McGinty Creek. Generally,

concentrations were higher in April/May likely due to mobilization by the elevated DOC levels observed at this time of the year;

- Dissolved molybdenum concentrations typically increased moving downstream in McGinty Creek and varied seasonally such that lower concentrations were observed in the spring and higher concentrations were observed in the winter, suggestive of dilution of baseflow providing the bulk of molybdenum to McGinty Creek in the spring;

- Dissolved lead and silver concentrations have generally been close to or below their respective detection levels in McGinty Creek;

- Dissolved selenium concentrations in McGinty Creek were generally highest in the west arm (reference tributary) at MN-0.5, suggesting a significant proportion of loading occurs from the west arm to McGinty Creek downstream of MN-2.5 and upstream of MN-4.5.

6 GROUNDWATER

Current groundwater conditions for the Minto Mine are presented in multiple reports. Groundwater conditions at the Minto site are monitored as part of the Minto Mine Groundwater Monitoring Plan (GMP, 2016-02), with monitoring data reviewed as part of the Operational Adaptive Management Plan (OAMP, 2017-02). The *Groundwater Characterization and Conceptual Model Update Report* (SRK 2018a) includes an assessment of data collected from groundwater wells in the area, results from the environmental tracer study and an assessment of potential transport of mine contact water across the site (see Appendix 1-4a). Updated conceptual models are presented for both the Minto Creek and McGinty Creek catchments. SRK (2018b) presents a summary of groundwater quality at OAMP monitoring stations that has reached thresholds defined in the OAMP (see Appendix 1-4b). The sum of this information is the basis for this section.

In the Minto Creek catchment, groundwater is fracture-controlled and, at the largest scale, groundwater flow follows topography, except where influenced by the Main and Area 2 open pits, and the Minto South Underground. These mine components create localized hydraulic sinks that overprint the topographic flow system. Locally, permafrost impedes downward infiltration of water from reaching the deeper groundwater system. Groundwater that does flow out of the mine footprint will move towards the Minto Creek valley and the Yukon River. Most of groundwater from the mine is expected to discharge to surface water by or within the Water Storage Pond. Any groundwater that does bypass the Water Storage Pond would be assumed to discharge to Minto Creek by surface water station MC1.

Seepage of contact groundwater from the open pits is interpreted to have remained within the mine footprint. OAMP events for groundwater quality are limited to SO_4, As, Cd and Se; other parameters have remained stable or not shown consistent changes sufficient to trigger the OAMP. As and Cd events have been inconsistent and, as of the last 2017 sampling round, did not exceed thresholds. Se exceedances have also been inconsistent and do not show consistent trends. SO_4 exceedances in groundwater at the mine property boundary are interpreted to not be related to mine seepage (Appendix 1-4a).

In the McGinty Creek catchment, the only current mine component is the Minto North open pit and related roadwork. In this catchment, groundwater is fracture-controlled and flow follows topography, with the Minto North open pit acting as a hydraulic sink; the open pit has not filled with water. Groundwater in this catchment that does not report to the open pit will move towards the Yukon River and is assumed to discharge to McGinty Creek.

There is interpreted to be no seepage of contact groundwater from the Minto North open pit as the pit is acting as a hydraulic sink. OAMP events for groundwater quality have occurred for As, Zn, Cd, NO_3 and SO_4. Zn and Cd events have been inconsistent. As and NO3 exceedances in groundwater have been consistent but are interpreted to be related to flushing of non-PAG haul road rock via precipitation. SO_4 exceedances of baseline concentrations have been ongoing, but concentrations are very low (<5 mg/L) and not considered significant.

7 WILDLIFE

Wildlife assessments were completed within, or near, the Project area between 1994 and 2017. This section summarizes the results of wildlife surveys conducted in the area to date, identifies species of concern, and lists ongoing and/or near-future surveys to be completed. The last update on the number and types of wildlife studies that have been conducted in the Minto Mine area was in a comprehensive 2010 report produced by EBA Engineering Consultants Ltd (EBA), titled *Minto Mine, Environment Baseline Report – Wildlife*.

7.1 ENVIRONMENTAL SETTING

The Minto Mine is located within the Boreal Cordillera ecozone and in the western part of Yukon Plateau Central ecoregion (Smith et al., 2004). The Minto Mine is situated in the far western part of the Yukon Plateau-Central ecoregion near the Dawson Range and adjacent to the Klondike Plateau ecoregion in the west. The area was part of the eastern extent of Beringia, which remained ice-free approximately twenty to fifteen thousand years ago during the last maximum glacial period (Smith et al., 2004).

The Minto Mine is in the eastern part of the Dawson Range the local elevation ranges from 700 m to 950 m; the general landscape is composed of rounded mountains intersected by broad valleys and drainages that flow into the Yukon River. The access road starts on the western side of the Yukon River, at the barge landing site, continues north along the Yukon River and then turns southwest up the Minto Creek valley for 12 km to reach the Minto Mine site.

Forest fires are frequent in this region as less than 300 mm of precipitation falls per year due to the rain shadow formed by the St. Elias-Coast Mountains in the west. As a result, the study area around Minto Mine has experienced numerous fires over the last thirty years, rendering it into a complex mosaic of plant communities at various stages of succession. The oldest pertinent fire burned approximately 7,236 ha in 1980, the second and more extensive fire occurred in 1995 and burned approximately 55,521 ha (GYWFM, 2012). The 1995 burn occurred along the access road by the Yukon River barge landing and just west of the airstrip as shown in . The most recent fires occurred in 2010 and 2011, these small fires only consumed 17 hectares within the Minto claims, southeast of the airstrip.

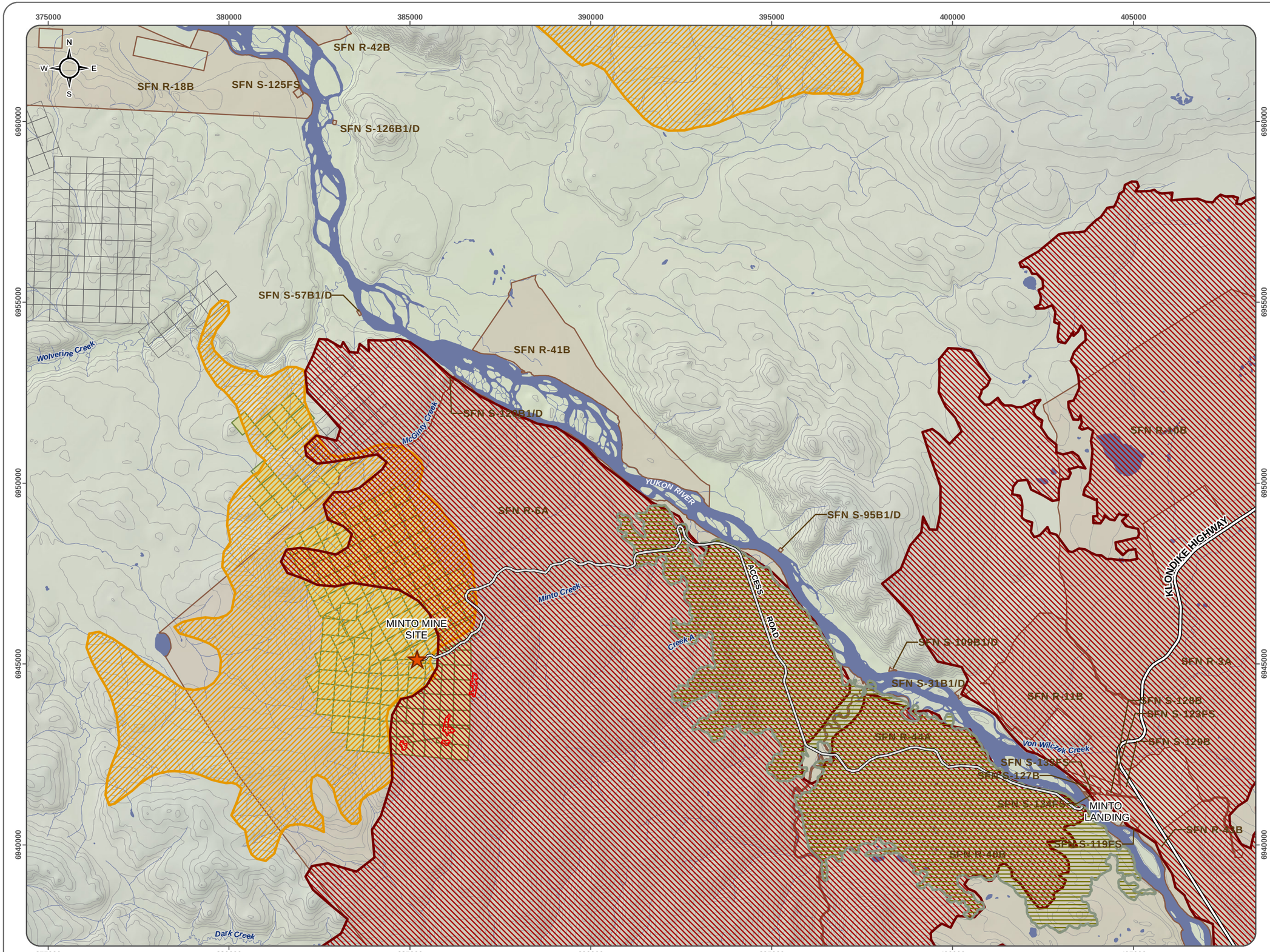
The fire-disturbed areas are now regenerating and young forest or shrub ecosystems dominate the Minto area (Oswald and Brown, 1990). Willows (*Salix sp*.) and trembling aspen (*Populus tremuloides*) are the most represented species in crown cover at present. Lodge pole pine (*Pinus contorta latifolia*) is a later successional species and will gradually dominate well-drained mid and upper slopes. Shade-tolerant white spruce is the regenerating climax tree species currently found in the understory as seedlings. White spruce will eventually overgrow the pine and trembling aspen communities. Black spruce (*Picea mariana*) is also a climax species that is adapted to wetter, cooler sites, and is often the persistent species in white/black spruce-mixed areas along slope toes, valley bottoms and northern aspects. Small grasslands are scattered along dry crests and steep south-facing slopes, these locations do not retain enough moisture to sustain tree growth.

MINTO MINE

CAPSTONE MINTO MINE - SITE CHARACTERIZATION PLAN

FIGURE 7-1
FIRE HISTORY

JANUARY 2018



Minto Mine Site

Control Burn 2010

1980 Fire

1995 Fire

2010 Fire

Selkirk First Nation Settlement Lands

MInto Exploration Ltd. Claims

Other Claims

Mine Access Road

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0 1 2 3 4 5

kelometres

First Nation Settlement land obtained from Natural Resource Canada. Quartz claim boundaires and ownership are current as of 12 January 2018. Data obtained from Mineral Resources Branch, Energy Mines and Resources Department, Government of Yukon.
Site hydrology data provided by Minto Explorations Ltd, May 2009.

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The diversity of vegetative communities and successional stages around the Minto Mine provides a variety of habitat niches that support approximately 46 species of mammal (insectivores, bats, lagomorphs, rodents, carnivores, and ungulates), 60 species of birds, and one species of amphibian, the wood frog (*Rana sylvatica*). The list of mammals and bird species known to exist in the Minto area and/or the Yukon Plateau-Central ecoregion is included in EBA's 2010 baseline report.

7.2 WILDLIFE BASELINE ASSESSMENTS

lists the wildlife surveys and studies that have been conducted since 1994 in the Minto Mine area.

Table 7-1: Wildlife Surveys and Studies Undertaken in the Minto Project Area.

Dates	Type of Survey	Conducted By
Jan-March 2012	Late Winter Ungulate Studies	EDI, Environment Yukon
Fall 2012	Klaza Caribou Herd Study	Environment Yukon
March 2011	Late Winter Ungulate Study	Environment Yukon and EDI (on behalf of Casino Mining Corporation)
July 2010	Baseline Ecosystems and Vegetation Survey	Access Consulting Group
February 2010	Late Winter Moose Survey (Aerial)	Access Consulting Group
December 2010	Post-rut Moose Survey (Aerial)	Access Consulting Group
June 2009	Dall Sheep Survey (Aerial)	Environment Yukon
2007	Moose Survey	Environment Yukon
2003	Klaza Caribou Herd Survey	Environment Yukon
1994	Spring Wildlife Survey Spring Dall Sheep Survey Summer Raptor Survey Summer Wildlife Ground Pellet Survey	Hallam Knight Piesold Ltd.

The three most recent wildlife baseline studies were led by the Yukon Government (YG) and were not designed specifically for the Minto Mine project. These studies encompassed the large Carmacks West Moose Management Unit and the Klaza caribou herd range, for a total area of 6,430 km^2 which overlapped the Minto Mine site. The results from the YG surveys, when released, will provide more statistically sound estimates of moose and caribou population levels, gender ratios and recruitment success for the overall region. These values can then be used as means of gaging how Minto Mine area resident ungulate populations are faring. Also, the locations of sightings when mapped show distribution and can help detect high value habitat that may require protection if mine and road development increases substantially beyond the current footprint (which is not currently anticipated).

The 2010 Baseline Ecosystems and Vegetation Study (Access Consulting Group (ACG), 2010) was not primarily focused on wildlife; however, general wildlife observations were made during the vegetation survey, recorded on plot data sheets and wildlife mitigation recommendations were included in the report.

A wildlife log is maintained on site to track the animal sightings, details and behaviour in the site vicinity, and to monitor any human-wildlife interactions.

7.3 WILDLIFE SPECIES OF CONSERVATION CONCERN IN YUKON

The Minto Mine area and surrounding environment provide habitat for several species considered at risk by both the federal and territorial governments. lists species that have been assessed by the Committee on the Status of Endangered Wildlife in Canada (COSEWIC) and/or the Yukon Government as needing special attention and protective legislation, so that remaining populations are not unduly stressed. The list consists of species whose range overlaps the Minto Mine area.

Table 7-2: Wildlife Species of Conservation Concern in Yukon.

Species	Status*	Source
Little Brown Bat (*Myotis lucifugus)*	Endangered	COSEWIC (2012)
Common Nighthawk *(Chordeiles minor)*	Threatened	COSEWIC (2007)
Olive-sided Fly Catcher *(Contopus cooperi)*	Threatened	COSEWIC (2007)
Barn Swallow (*Hirundo rustica)*	Threatened	COSEWIC (2011)
Canada Warbler *(Wilsonia canadensis)*	Threatened	COSEWIC (2008)
Peregrine Falcon (*Falco peregrinus anatum - tundrius)*	Special concern	COSEWIC (2007), Yukon Wildlife Act (2002)
Short Eared Owl *(Asio flammeus)*	Special concern	COSEWIC (2008)
Wolverine *(Gulo gulo)*	Special concern	COSEWIC (2014)
Grizzly Bear *(Ursus arctos)*	Special concern	COSEWIC (2012)
Northern mountain population of woodlandCaribou *(Rangifer tarandus caribou)*	Special concern	COSEWIC (2014), SARA (2002)
Collared Pika *(Ochotona collaris)*	Special concern	COSEWIC (2011)
Rusty Blackbird *(Euphagus carolinus)*	Special concern	COSEWIC (2017)
Gyrfalcon (*Falco rusticolus)*	Specially protected	Yukon Wildlife Act (2002)
Trumpeter Swan *(Cygnus buccinator)*	Specially protected	Yukon Wildlife Act (2002)
Mule Deer *(Odocoileus hemionus)*	Specially protected	Yukon Wildlife Act (2002)
Cougar *(Puma concolor)*	Specially protected	Yukon Wildlife Act (2002)
Bank Swallow (*Riparia riparia*)	Conservation concern	Yukon Environment (2011)
Northern Shrike (Lanius excubitor)	Conservation concern	Yukon Environment (2011)
Mountain Goat *(Oreamnos americanus)*	Conservation concern	Yukon Environment (2011)

* Status designations by COSEWIC are defined as:

Extinct - A wildlife species that no longer exists.

Extirpated - A wildlife species that no longer exists in the wild in Canada, but exists elsewhere.

Endangered - A wildlife species facing imminent extirpation or extinction.
Threatened - A wildlife species that may become endangered if factors leading to its extirpation or extinction are not reversed.
Special Concern - A wildlife species that may become threatened or endangered because of its biological characteristics combined with environmental impacts.
The Yukon wildlife Act provides more legal protection for those wildlife species recognized as Specially Protected. Yukon Environment wildlife species of Conservation Concern are populations that are decreasing and require more monitoring.

There are currently eleven wildlife species in Yukon rated (in 2017) as threatened or of special concern by COSEWIC, of which ten species have ranges that could possibly encroach upon the area around Minto Mine. These include: grizzly bear (*Ursus arctos*), northern mountain woodland caribou (*Rangifer tarandus caribou*), wolverine (*Gulo gulo*), collared pika (*Ochotona collaris*), bank swallow (*Riparia riparia*), barn swallow (*Hirundo rustica*), rusty blackbird (*Euphagus carolinus*), olive-sided fly catcher (*Contopus cooperi*), common nighthawk (*Chordeiles minor*), short-eared owl (*Asio flammeus*), and peregrine falcon (*Falco peregrines anatum*).

One species, the little brown bat (*Myotis lucifugus*), has been classified by COSEWIC as an endangered species, which means they are in immediate danger of extinction or extirpation. Populations of these mammals have been seriously affected by a fungal infection known as White Nose Syndrome. The Minto Mine is within the northern extent of the summer range of the Little Brown Bat, although their existence within the area has not been verified.

The following sections provide brief descriptions of the most recent wildlife surveys conducted locally for Minto and surveys conducted by the Yukon Government that were larger in scale and included the Minto Mine area.

7.3.1 Moose

Current information regarding moose in the area came from two aerial surveys conducted by Environment Yukon and Environmental Dynamics Inc. (EDI) that covered a survey area that included the Minto Mine site. The average density of moose in the Yukon typically ranges from 100 to 250 moose/1,000km^2 (Yukon Government, 2017a). Estimates of moose in the Minto Mine area have ranged from 20 to 127 moose/1,000km^2 (EDI, 2013).

7.3.1.1 Aerial Moose Survey – Late Winter 2011/2012

This late winter aerial survey covered 6,400 km^2 west and northwest of Carmacks. During this survey, a total of 311 moose were observed. The 2011 map shows a cluster of moose observation points around the Minto Mine and in the southeast highlands, indicating that this area has a high density of moose in the late winter as compared to most of the Regional Survey Area (G. Pelchat, pers. comm.). Minto Mine vicinity has been subjected to numerous fires in the last thirty years and the vegetation cover is dominated by tall shrubs, attractive habitat for moose with plenty of browse and cover.

In addition, the Yukon River is approximately 10 km to the east. This large river corridor has ample shoreline and islands for moose calving and post-calving habitat. The Ingersoll Islands, located in the Yukon River downstream of the project site, are known to be used for calving during the spring and as rearing habitat during the summer (Magrum, 1994). The old burn areas in the Minto Creek Valley, the banks of the Yukon River, and the swamp lands below Minto Creek are often used by moose during the spring and summer (EBA, 2010b).

7.3.1.2 Aerial Moose Survey - Winter 2009/2010

Aerial moose surveys were completed on December 15, 2009 (post-rut) and February 23, 2010 (late winter) by ACG (2010). The total area surveyed was 112 km^2, specifically concentrating on the area around the Minto Mine and nearby drainages, see . Moose density for the post-rut survey was estimated to be 125 moose per 1,000 km^2. The average population density for calf-to-cow ratio estimated from this data was 25 calves and no sub adults for every 100 adult cows, and the estimated adult sex ratio was 50 mature bulls for 100 cows, which is considered fair compared to territorial averages.

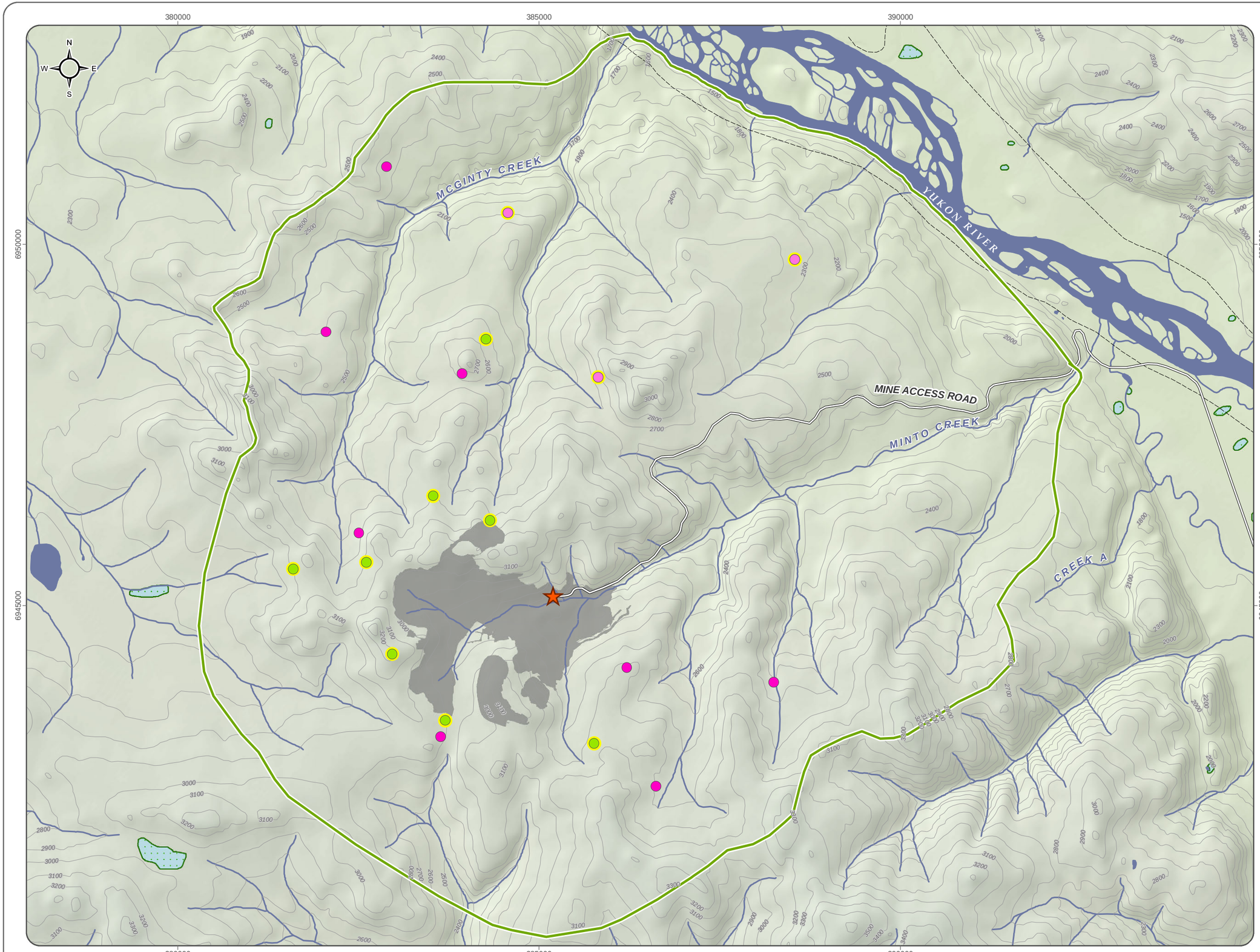
MINTO MINE

CAPSTONE MINTO MINE - SITE CHARACTERIZATION PLAN

FIGURE 7-2

WINTER 2009/2010 AERIAL MOOSE SURVEY

JANUARY 2018



- Moose Observation (15-Dec-2009)
- Moose Observation (23-Feb-2010)
- Moose Tracks (23-Feb-2010)
- Winter 2009/2010 Survey Boundary
- Wetland
- Minto Access Road
- Contours (100ft)

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7.3.2 2007 Early Winter Moose Survey

The early winter 2007 moose survey for the Carmacks West Moose Management Unit (MMU) was conducted by Yukon Department of Environment. The densities are comparable, as there are similar habitat types in the MMU and its boundaries overlap the Minto Mine. The calculated moose density was 124 moose per 1,000 km^2 for the survey area.

This survey covered a much larger study area than the surveys conducted by ACG during the winter of 2009/2010, which were specifically focused on the area surrounding the Minto Mine site. However, during this survey in 2007, a total of 208 moose were observed during the survey, with a total population estimate of 520 moose for the study area. Survival rates for calves and yearlings were relatively low. The sex ratio of 75 bulls per 100 cows is considered to be a healthy sex ratio. The average sex ratio for other areas surveyed within Yukon is 68 bulls per 100 cows (O'Donoghue et al., 2008).

7.3.3 1994 Late Winter Moose Survey

Wildlife Resources surveys conducted in January 1994 by the Yukon Government estimated a moose density of 40 moose per 1,000 km^2 in the Minto area, which was lower then the 100 to 250 per 1,000 km^2 Yukon average (Yukon Government, 2017).

7.3.4 Caribou

The closest Woodland caribou ranges to the Minto Mine belong to the Klaza and the Tatchun herds. The Klaza herd appears to be the more stable of the two, and is the larger herd. Below is a brief summary of the herds' interactions with the Minto Mine area.

In 2005, the Klaza herd population was estimated at 650 and predicted to increase (Yukon Environment, 2005a). There are concerns for this caribou herd, as an increase in exploration projects and road development may cause negative impacts to the health of the Klaza caribou. The Klaza caribou range is west of the Minto Mine. As the area around the Minto Mine has experienced numerous fires recently, the habitat is of minimal value for caribou (Hegel, pers. comm.). Caribou prefer mature open forests where arboreal and ground lichen are plentiful. A Wildlife Key Area (WKA) for woodland caribou winter range was identified approximately 9 km to the east-northeast of the project area (Yukon Environment, 2010a). A fall (rut) count of Klaza caribou herd was conducted by Environment Yukon, EDI and Little Salmon Carmacks First Nation members in 2012. The herd is currently estimated to be approximately 1,180 caribou and is considered stable (Francis et al., 2016).

The Tatchun Caribou herd range is to the east of the Yukon River and does not overlap with the Minto Mine. In 2012, the population estimate for the Tachun herd was 500 animals.

MINTO MINE

CAPSTONE MINTO MINE - SITE CHARACTERIZATION PLAN

FIGURE 7-3

WILDLIFE KEY AREAS

JANUARY 2018

capstone MINTO MINE

WILDLIFE KEY AREAS

- Golden Eagle
- Peregrine Falcon
- Bald Eagle
- Alpine Raptor
- Thinhorn Sheep
- Woodland Caribou

OTHER MAP FEATURES

- Minto Mine Site
- Mine Footprint
- Wetland
- Minto Access Road
- Contours (100ft)

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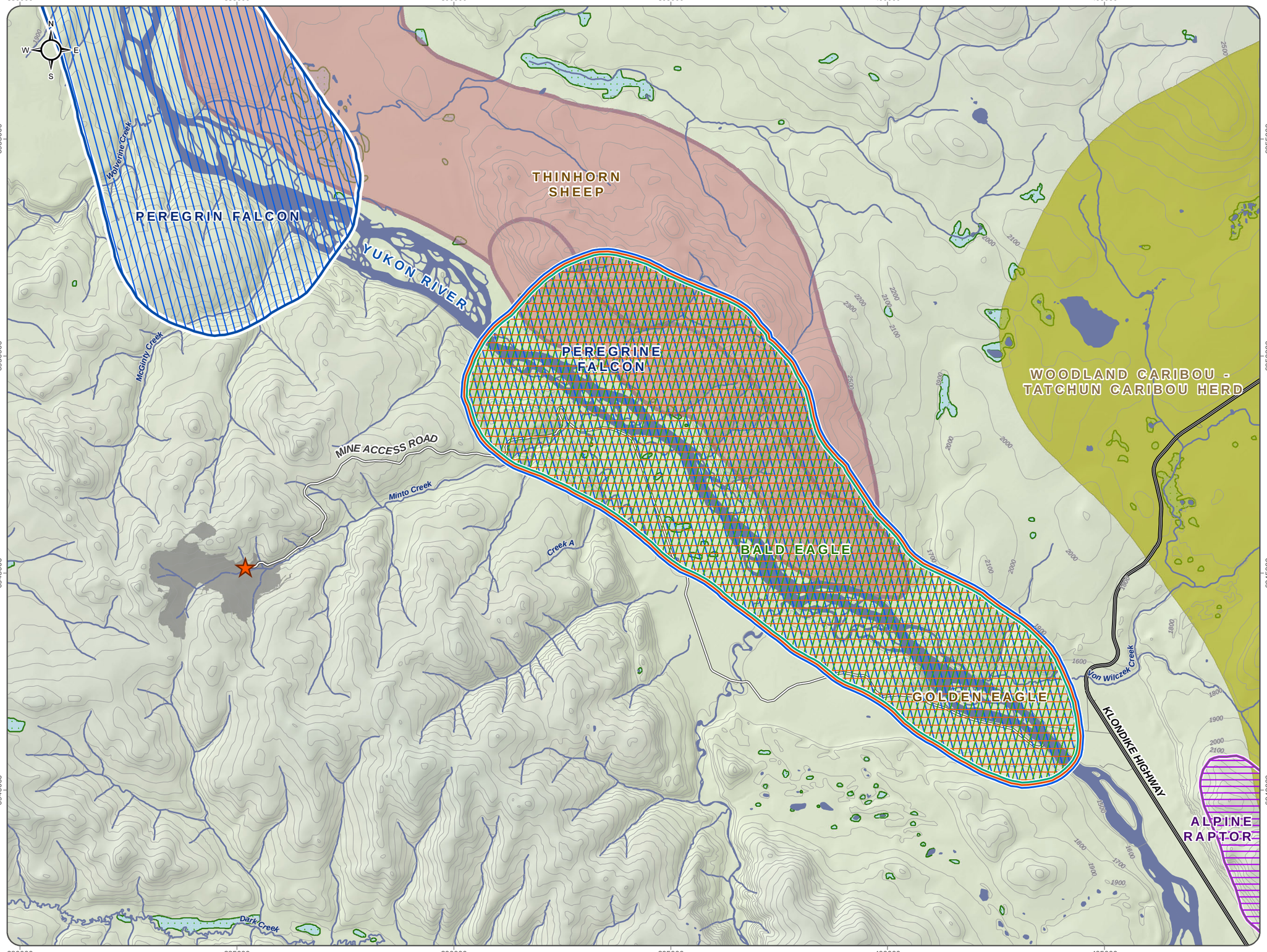
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7.3.5 Sheep

The Minto Bluffs along the Yukon River have been identified as an important Dall sheep area, see Figure 2, (O'Donoghue, 2009). Although the access road to the Minto Mine passes near sheep habitat, sheep habitat within the project area itself is limited and sheep are not expected to inhabit the project area for any extended length of time.

Between 2000 and 2017, sheep surveys of the Minto Bluffs resulted in observations of between 31 and 153 sheep annually, with the majority of observations being ewes, yearlings and lambs. During the 2016 survey, 153 sheep were observed, of which 59 were observed on the Minto Bluffs (which is located about 8 kilometres downstream and across the river from the Minto Mine site). This is the highest recorded observation for this area. Most sheep observed during these surveys have been located on the Minto Bluffs, Split Mountain, and Mount Hansen (O'Donoghue, 2009).

7.3.6 Carnivores, Fur Bearers and Small Mammals

Fur trapping and big game harvest statistics indicated the following species are expected to occur in the Minto Mine area: grizzly bear, black bear, coyote, gray wolf, red fox, wolverine, marten, least weasel, beaver, and lynx. Cougars may also have the potential to be found in the area as they are known to follow mule deer (Smith et al. 2004). Of the species listed above, the following species (or their sign) have been observed, on site: grizzly bear, black bear, gray wolf, lynx, river otter (HKP 1994, Capstone 2007-2008).

The territorial estimates for bear populations in Yukon are 6,000 to 7,000 grizzly bear and 10,000 black bear (Yukon Environment 2010b). The Yukon Government (2017b) has indicated that key habitat for black bears include seasonally concentrated feeding areas, such as south aspect slopes containing sagewort, bearberry, and grassland habitat. Summer and fall feeding habitats typically consist of those areas where berries grow. Black bears have been observed in the project area on many occasions (Wildlife log – Appendix 1-5).

Grizzly bears are known to use the Minto Mine area. Sightings and tracks were documented in the 2010 Baseline Ecosystems and Vegetation Report, and sightings have been reported by mine personnel (Appendix 1-5). Key habitat for grizzly bears includes areas where they concentrate seasonally, such as feeding areas, floodplains, and movement corridors. Important feeding habitat includes areas with profuse berries and areas where salmon spawn (Yukon Environment, 2017b). Although key habitat for grizzly bears has not been identified on site, they have been seen in the project area regularly since the commencement of mine operations. (ACG, 2010).

Wolverines have large territories where they hunt, scavenge and mate. While there is a moderate probability that they inhabit the area, because of their aversion to human activity and low population, they are not likely to be observed.

Small mammals common to the area include red squirrel, varying hare, fox, mink, weasel, vole, and shrew. The Minto Mine site is situated at the apex of five drainages that are part of the Yukon River watershed, so wildlife uses the area to access the valleys offering conduits from lowlands to highlands for seasonal foraging and hunting (ACG, 2010).

7.3.7 Birds

Five species of birds are considered to be of conservation concern: the peregrine falcon, short-eared owl, common nighthawk, olive-sided flycatcher, and rusty blackbird. Three of these species, the peregrine falcon, common nighthawk, and olive-sided flycatcher, have a moderate probability of occurrence at the Minto site. Suitable nesting habitat for the peregrine falcon is located in close proximity to the project site (see Figure 7-3), on the bluffs along the Yukon River (O'Donoghue, pers. comm.), and a historical record of nesting for this species was documented at the Pelly–Yukon River confluence (Mossop, pers. comm. as cited in HKP 1994). Common nighthawks are often found near open lodge pole pine forests, old burn areas and open mixed forests, and near wetlands or rivers (Sinclair et al., 2003) and many of these habitat types occur in the project area. The olive-sided flycatcher often occurs in black and white spruce, lodge pole pine, and mixed forests, from lowland areas to tree-line. The short-eared owl and rusty blackbird are considered to have a low probability of occurrence in the project area. The short-eared owl is often associated with open wetland and meadow, alpine, and alpine tundra habitat, which is limited within the project area. The rusty blackbird is also associated with wetland habitat (Sinclair et al., 2003), which is not abundant within the project area itself, but may occur along the margins of the Yukon River.

The rusty blackbird often nests at the edge of ponds/wetland complexes in boreal forests. They prefer lower elevations, but could feed and nest within the study area. Short-eared owls may travel through the area, but their typical habitat is in and near large meadows and agricultural fields and they may not be seen often because they are nocturnal. Peregrine falcons do not use the habitats in the Minto area; they nest and hunt near steep canyon walls along the Yukon River, approximately 20 km east.

7.3.7.1 Raptors

Although numerous raptor species have the potential to inhabit the project area, species that have been observed and documented in the Minto Mine area include the red-tailed hawk (HKP, 1994), peregrine falcon (Mossop, pers. comm. as cited in HKP 1994), and golden eagle (O'Donoghue, pers. comm.). It should be noted that only one aerial-based raptor survey was conducted as part of the Minto Mine baseline studies (HKP, 1994). High quality riparian cliff habitat for raptors exists along the Yukon River downstream of the Minto Mine access road. A WKA (wildlife key area) for golden eagle summer nesting habitat has been identified approximately 3 km to the east of the project area (Yukon Environment, 2010a) (Figure 7-3). This WKA is primarily associated with the steep bluffs along the Yukon River and includes a buffer area. No cliff-nesting raptor habitat has been identified within the Project area itself. The access road to

the Minto Mine, however, runs adjacent to potential nesting areas for cliff-nesting raptors, such as the golden eagle and peregrine falcon.

7.3.7.2 Waterfowl

Key habitat for waterfowl includes wetlands that are used as staging areas in the spring and fall, and for breeding and molting in the summer. As suitably-sized wetlands are not found near the Minto Mine, waterfowl are not known to be present in large numbers or for extended periods. A key habitat area, Lhatsaw wetlands, lies approximately 30 km east of the project site and is used for nesting and molting in the summer (Yukon Environment, 2016). Waterfowl that may occasionally use the Minto Creek drainage include Canada goose, mallard, northern pintail, green-winged teal and American widgeon.

7.3.7.3 Game Species

Game birds that have been observed or that have the potential to occur in the study area include grouse (spruce, ruffed, sharp-tailed) and ptarmigan (willow, white-tailed, and rock). Of the species of grouse that live in Yukon, the sharp-tailed grouse is currently the only species of management concern. Sharp-tailed grouse have a limited distribution in Yukon due to the lack of suitable habitat. Gravel outwashes with fairly stable aspen parkland habitat and wet sedge-hummock meadows after fire are considered suitable habitat for this species. Sharp-tailed grouse have been observed in the Project area.

7.3.7.4 Amphibians

Of the five amphibian species known to occur in Yukon and northern British Columbia, only the wood frog (*Rana sylvatica*) is known to occur in the project area. This species is restricted to wetland areas and has not been surveyed for at the Minto site, although there have been some anecdotal observations. No reptiles are known to occur in Yukon.

7.4 SUMMARY

The frequent and often large fires that have occurred around the Minto Mine have created high quality habitat for moose. The numerous moose sightings and signs found in the area indicate that it is attractive and well-used by resident moose. Local aerial surveys completed in early (December 2009) and late winter (February 2010) indicated that the moose population in the area was below territorial average and recruitment may be low. However, the population has increased since the initial local survey was done in 1994. Then, the population was estimated at 40 moose per 1,000 km^2 during a government-supported count. Winter surveys in 2007 indicated a population of 125 per 1,000 km^2.

The Klaza woodland caribou herd range is located approximately 10 km west of the Minto Mine, where there is better winter habitat and mature open forests with ample growth of arboreal and ground lichens. The Minto Mine area does not provide good habitat for caribou, but they may travel through occasionally.

Current wildlife population estimates for caribou and moose conducted by Environment Yukon biologists are done over large regional study areas. Average population statistics, especially for moose, may not accurately reflect local habitat usage or seasonal variations, but can serve as a background comparison.

Dall sheep habitat is found along the steeper hillsides within the Yukon River corridor; the Minto Mine area does not contain the steep escape topography nor grasslands needed by these animals. The nearest population of Dall sheep is found at the Minto Bluffs approximately 10 km further down the Yukon River. The nearest disturbance is the access road, which is still approximately 10 km from sheep habitat.

Raptor inventories have been concentrated along the Yukon River corridor where steep slopes and cliffs provide nesting and perches for peregrine and golden eagles. Bald eagles commonly use large trees for nesting and are closely associated with riparian systems. Forest-dependent raptors such as northern goshawk and red-tailed hawks are likely nesting around the Minto Mine in mature forest areas.

Black and grizzly bears are known to use the area and may be attracted to food smells emanating from the camp and garbage disposal area. A waste management plan is in place to reduce attracting bears to camp and operation areas.

Large mammals such as moose, mule deer, timber wolves, grizzly bears, and black bears were found to be using riparian corridors, secondary access roads, and exploration transects as migration routes throughout the Minto mine study area (ACG 2010).

There is a dearth of information on other animals existing in the vicinity of Minto such as wolves, lynx, mule deer and bears.

8 AQUATIC RESOURCES

8.1 OVERVIEW

A complete report on Aquatic Resources is presented in the *Minto Mine Site Characterization – Aquatic Resources (AEG and Minnow, 2017)* (Appendix 1-6), which constitutes an update to the 2013 Aquatic Resources Baseline Report which was compiled by Access Consulting Group (ACG) and Minnow Environmental as part of the Phase V/VI Yukon Environmental and Socio-Economic Assessment Board (YESAB) Project Proposal. A summary of the key findings of the 2017 report is presented in the sections below.

The Minto Mine region includes the Yukon River and its smaller tributaries, including 7 km upstream to Big Creek and 13 km downstream to Wolverine Creek. The local study area of the Minto Mine centres on four small drainages that drain directly into the Yukon River: Minto Creek, Big Creek, an unnamed creek (referred to as Creek A) and McGinty Creek. The primary drainage is that of Minto Creek, which flows northeast from the existing mine site over approximately 17 km to the Yukon River, and covers an area of roughly 41 km^2. Big Creek flows to the northeast over about 97 km and is crossed by the Minto Project access road. Creek A flows to the north over about 7 km near the lower end of Minto Creek, drains an area of roughly 9 km^2, and is crossed by the Minto Project access road. The fourth drainage is that of McGinty Creek (formerly referred to as Unnamed Creek B), which is located to the north of Minto Creek and flows north-northeast nearly 9.5 km to the Yukon River confluence, and covers an area of about 34 km^2. Operational data was collected for Minto Mine's Annual WUL and through 3 Cycles of EEM.

8.2 FISH AND FISH HABITAT

8.2.1 Regional Overview (Yukon River)

A variety of resident and migratory fish species inhabit the Yukon River near Minto Mine. These include Chinook, Coho and chum salmon, lake trout, least cisco, Bering cisco, round whitefish, lake whitefish, inconnu, Arctic grayling, northern pike, burbot, longnose sucker, and slimy sculpin.

Previous studies on the Yukon River within the vicinity of Minto Mine have identified both spawning and rearing areas for salmon. Spawning shoals are present in the Ingersoll Islands (downstream of the Project area) as well as around islands upstream of Minto Mine, near Big Creek. These offer an extensive network of side channels and sloughs which provide good spawning gravel.

This portion of the Yukon River also provides rearing habitat for Chinook salmon, as evidenced by past studies in the Project area. Juvenile Chinook salmon (JCS) generally spend up to 1.5 years feeding and growing in fresh water tributaries prior to out-migrating to the ocean, and feed or stage in various tributaries to the Yukon River during this slow out-migration. Usage of the Project area tributaries by JCS is outlined further below.

8.2.2 Local Fish Habitat Investigations

8.2.2.1 Minto Creek

Fish and fish habitat studies of Minto Creek have been ongoing for many years, with contemporary studies including those from 1994 through to 2017 (see Table 8-1). A summary of effort and catches is presented in Table 8-2 while complete results can be found in Appendix 1-6. Generally, Minto Creek has been noted to provide only limited habitat to fish. Flows within the stream are quite variable on a yearly basis, with intermittent flows and extensive ice build-up during winter that limits the potential for overwintering habitat for fish. Also, the distribution of fish within Minto Creek has been observed to be limited to the lower 1.5 km of the watercourse, as there is a barrier and steep canyon upstream of that location. As noted above, Chinook salmon, slimy sculpin, Arctic grayling, longnose sucker, burbot, and round whitefish have been captured in Minto Creek; however, the latter have not been observed since the original baseline studies in 1994. Slimy sculpin have been observed consistently, but at a low density.

Table 8-1: Summary of Key Fish and Fish Habitat Studies, Minto Mine.

Year	Firm and Study Name	Scope of Studies
1994	Hallam Knight Piesold (HKP) – IEE for Minto Project Area (HKP 1994)	• Fisheries investigations on Minto Creek and Creek A. • Backpack electrofishing, minnow trapping. • Reach definition and description, identification of barriers to fish passage.
2006–2007	Access Consulting Group (ACG), R&D Environmental – Various Fisheries Investigations for Minto Explorations Ltd. (Minnow/ACG, 2007)	• Fisheries investigations in Minto Creek to support the permitting of the Minto Mine. • Backpack electrofishing, minnow trapping.
2008	ACG, Minnow Environmental – EEM Program, Cycle 1 (Minnow/ACG, 2009)	• Fisheries investigation of Minto Creek. • Backpack electrofishing and minnow trapping.
2009	ACG – Fish Relocation Program (ACG, 2009)	• Minnow trapping in Minto Creek and transfer of fish to the Yukon River.
2010	ACG – Fish Mark and Recapture Program (ACG, 2010)	• Minnow trapping in Minto Creek and marking of captured fish (release back into Minto Creek).
2009–2011	ACG, Minnow Environmental – EEM Program, Cycle 2 (Minnow/ACG, 2012)	• Integrated assessment of effluent sub-lethal toxicity, water quality, benthic invertebrate community condition, and fish health (hatchery-based exposure study).
2011–2012	ACG – Minto Creek Fisheries Monitoring Program (ACG, 2012/ ACG, 2013)	• Minnow trapping in Minto Creek.
2009–2011	ACG – Fisheries Monitoring Program in McGinty Creek	• Fisheries Investigation of McGinty Creek through Minnow trapping.
2012	ACG- Fisheries Monitoring in Minto Creek and Big Creek	• Fisheries investigation of Minto Creek and Big Creek through minnow trapping and backpack electrofishing
2013	ACG- Fisheries Monitoring in Minto Creek and Big Creek, EEM Program Cycle 3	• Fisheries investigation of Minto Creek and Big Creek through minnow trapping and backpack electrofishing

Year	Firm and Study Name	Scope of Studies
2014	ACG- Fisheries Monitoring in Minto Creek and Big Creek	• Fisheries investigation of Minto Creek and Big Creek through minnow trapping.
2015	ACG- Fisheries Monitoring in Minto Creek and Big Creek	• Fisheries investigation of Minto Creek and Big Creek through minnow trapping.
2016	Capstone- Fisheries Monitoring in Minto Creek and Big Creek	• Fisheries investigation of Minto Creek and Big Creek through minnow trapping.

Table 8-2: Summary of Fish captures in Minto Creek between 2008 and 2016.

Year	Method	Effort	Summary Statistics	Units	Juvenile Chinook Salmon	All Other Species
2008	Backpack Electrofishing	796 s	Catch	#	1	0
			CPUE	Fish/min	0.075	0
	Baited Gee Minnow Trapping	28.6 days	Catch	#	18	0
			CPUE	Fish/day	0.63	0
2009	Baited Gee Minnow Trapping	28.6 days	Catch	#	136	142
			CPUE	Fish/day	4.76	4.97
2010	Baited Gee Minnow Trapping	145.9 days	Catch	#	2293	2307
			CPUE	Fish/day	15.72	15.81
2011	Baited Gee Minnow Trapping	71 days	Catch	#	12	29
			CPUE	Fish/day	0.17	0.41
2012	Backpack Electrofishing	1051 s	Catch	#	0	4
			CPUE	Fish/min	0	0.23
	Baited Gee Minnow Trapping	43 days	Catch	#	3	6
			CPUE	Fish/day	0.07	0.14
2013	Backpack Electrofishing	3402 s	Catch	#	0	4
			CPUE	Fish/min	0	0.07
	Baited Gee Minnow Trapping	62.5 days	Catch	#	121	7
			CPUE	Fish/day	1.94	0.11
2014	Baited Gee Minnow Trapping	70.2 days	Catch	#	151	3
			CPUE	Fish/day	2.15	0.04
2015	Baited Gee Minnow Trapping	67.3 days	Catch	#	6	0
			CPUE	Fish/day	0.089	0
2016	Baited Gee Minnow Trapping	69.4 days	Catch	#	5	1
			CPUE	Fish/day	0.072	0.014

JCS presence in Minto creek was observed to be influenced by mine water discharge. During non-discharge periods, and prior to operations (baseline), numbers of JCS and other species of fish in the system were found to be very low. In contrast, during the two major discharge events (2009 and 2010), numbers of JCS were 20 to 30 times higher in the system (see Table 8-3 below), indicating that they may have been attracted into the system as a likely result of a more consistent temperature and flow regime associated with mine water discharge. Numbers of other species of fish during discharge events did not increase. It is important to note however that numbers of other species using the system has been consistently very low prior to mine development and during operations.

Table 8-3: Maximum Catch per Unit Effort (CPUE) of Juvenile Chinook Salmon in Minto Creek using baited Minnow traps 1994-2016.

Year	Month	JCS – CPUE (#fish/trap-day)*	Minto Creek Conditions
1994	September	0	Pre-development – no discharge
2008	September	0.9	Operational – no discharge
2009	September/October	20.0	Discharge
2010	August	30.0	Discharge
2011	September	0.43	No discharge – high TSS contribution from tributary
2012	September	0.19	No discharge – high TSS contribution from tributary
2013	October	5.01	Operational – no discharge
2014	September	5.05	Operational – no discharge
2015	July	0.30	Operational – no discharge
2016	July	0.301	Operational – no discharge

**CPUE calculated for actual 24-hr period (rather than nominal 24-hr period)*

Minto Creek does not provide preferred spawning habitat for fish because it completely freezes during winter months with no flow in lower Minto Creek. Accordingly, there is no evidence of spawning in Minto Creek (HKP 1994; R&D 2006, 2007), nor is there traditional knowledge indicating spawning occurring in the system (HKP 1994). Lower Minto Creek is also subject to low or zero flow conditions during periods in the summer when a portion (or all) of the flow sometimes infiltrates the ground, preventing the establishment of resident fish populations in this section of the stream. A natural barrier to fish passage exists at approximately km 1.2 upstream in Minto Creek from the Yukon River, therefore limiting fish use downstream of the barrier. This barrier is largely comprised of organic debris and can be considered temporary as it will likely degrade over time, although it has been consistently observed since it was first characterized in 2010. The canyon located upstream of the current barrier however is a permanent barrier to fish passage, due to its high gradient, thus limiting fish habitat to the lower 2.0 km of the system.

Water temperatures tend to remain cooler in Minto Creek than in the Yukon River until late June or early July. This likely deters fish, in particular JCS, from entering the system until creek temperatures equilibrate with the Yukon River. Minto Creek is subject to large diurnal fluctuations in temperature (up to 5°C or more) throughout the open water season. This daily variation is not ideal for fish and may limit their interest in using the creek for rearing.

Creek bottom substrate is comprised mostly of fines (silt/sand) with limited cobble/gravel sections which are more desirable for fish. In addition, significant input of suspended solids from a tributary in 2011 and 2012 may have further limited the use of the system by JCS during those years.

Bottom substrate at the mouth of Minto Creek and at the Yukon River confluence consists primarily of silt and mud which is not considered suitable substrate for salmonid spawning (grayling, salmon etc.). Aerial surveys for spawning fish were conducted in 2011, 2012, 2013 and 2014, and no adult fish were observed spawning in the vicinity of the Minto Creek/Yukon River confluence. No signs were present during the surveys that this area is used for spawning (such as redds or carcasses).

8.2.2.2 Big Creek

Starting in 2012, the compliance monitoring program included sampling efforts in Big Creek, which serves as a reference site. The same fish species as in Minto Creek were generally observed in Big Creek, and the CPUE has normally been slightly higher in Big Creek. Table 2-4 summarizes fish captures in Big Creek from 2012 to 2016.

Table 8-4: Summary of Fish captures in Big Creek in 2012 to 2016.

Year	Method	Effort	Summary Statistics	Units	Juvenile Chinook Salmon	All Other Species
2012	Backpack Electrofishing	273 s	Catch	#	1	23
			CPUE	Fish/min	0.22	5.05
	Baited Gee Minnow Trapping	11.8 days	Catch	#	7	2
			CPUE	Fish/day	0.59	0.17
2013	Backpack Electrofishing	911 s	Catch	#	0	27
			CPUE	Fish/min	0	1.78
	Baited Gee Minnow Trapping	14.8 days	Catch	#	19	2
			CPUE	Fish/day	1.28	0.14
2014	Baited Gee Minnow Trapping	16.4 days	Catch	#	96	3
			CPUE	Fish/day	5.86	0.18
2015		17.1 days	Catch	#	58	3

Year	Method	Effort	Summary Statistics	Units	Juvenile Chinook Salmon	All Other Species
	Baited Gee Minnow Trapping		CPUE	Fish/day	3.39	0.18
2016	Baited Gee Minnow Trapping	15.0 days	Catch	#	51	9
			CPUE	Fish/day	3.40	0.60

8.2.2.3 Creek A

Creek A was investigated during the 1994 baseline study program at the Project site, at which time no fish were observed or captured (including a site at the road-crossing location). Creek A is not considered to offer high quality habitat for fish.

8.2.2.4 McGinty Creek

Arctic grayling and slimy sculpin were captured in McGinty Creek in 1994, through electrofishing and minnow trapping. Because substantial deadfall caused by a forest fire changed creek conditions, only minnow trapping was used in 2009–2011, yielding very low numbers of slimy sculpin. Since these captures were consistently made in close proximity to the Yukon River, these fish were presumed to be associated with the Yukon River, as opposed to McGinty Creek. These results are similar to those found in the 1994 survey, in that fish were only captured in close proximity to the Yukon River confluence. The physical nature of the McGinty Creek drainage is not conducive to a consistent year-round use by fish. Many factors, including gradient, discharge volume, depth, configuration, and paucity of an upstream reservoir, limit wintering habitat potential for fish. Also, several potential natural fish barriers were observed and documented in the lower reach of McGinty Creek.

8.2.3 Aquatic Environment and Habitat

8.2.3.1 Stream Sediments

Stream sediments were studied for particle size and metal concentrations in 1994, and annually since 2006. Sediment particle size distribution was notably different when comparing earlier sampling years to more recent years. The change in distribution from 1994–2009 compared to 2010–2016 reflects methodological changes that were implemented in 2010. Sediment metal concentrations were also complicated by the change in methodology. With this qualification in mind, concentrations of arsenic, copper, and occasionally chromium exceeded the interim sediment quality guideline (ISQG) levels over the years but not greater than the probable effect level (PEL). Copper was the only metal that consistently exceeded guideline levels every year, including during baseline sampling in 1994. This could indicate that there are naturally high levels of copper at the exposure area. Due to the predominantly erosional habitat in upper Minto Creek, there are relatively few areas where sediment is deposited and then only in small quantities that likely wash away each year during freshet. Therefore, elevated sediment copper in the upper reaches of Minto Creek may be of limited importance in terms of exposure and potential toxicity to biota.

Two sediment toxicity tests, a 10-day *Chironomus dilutus* survival and growth, and a 14-day *Hyalella azteca* survival and growth test, were performed in 2011, 2015, and 2016. Lower Minto Creek sediment did not result in any effects to survival or growth of either test organism relative to the laboratory control or the field reference sediment (lower Wolverine Creek).

Additional detailed results for sediment physical and chemical parameters are available in Appendix 1-6.

8.2.3.2 Benthic Invertebrate Community

Benthic macroinvertebrates (benthos) are non-backboned animals inhabiting the bottom substrates of aquatic habitats. The abundance, diversity, and taxonomic composition of benthos can be used as indicators of changing environmental conditions as their distribution and abundance can be influenced by a wide variety of physical parameters. Baseline and numerous other benthic invertebrate studies were undertaken in the Minto Mine area from 2006–2016.

Basic results of the 2008 and 2011 environmental effects monitoring (EEM) benthic analyses indicated that Minto Creek (treatment) had a significantly higher benthic invertebrate density and slightly lower number of taxa (not significant) compared to McGinty Creek (reference). The 2011 EEM benthic results show that Minto Creek had significantly higher number of taxa and higher density compared to both reference sites. Increased taxa, higher density, and lower evenness are indicative of a site that is experiencing nutrient enrichment. During the EEM investigation of cause (IOC) study (Phase 3), the benthic invertebrate community of the mine-exposed sites on upper Minto Creek was compared to the

reference condition (defined by 12 reference areas) for four primary EEM metrics: density, taxon richness, Simpson's evenness and Bray-Curtis (BC) distance. Results are summarized in Appendix 1-6.

Under the terms of Minto's Water Use License #QZ06-006, benthic macroinvertebrate communities are required to be annually monitored in Minto Creek. Benthic invertebrate density at lower Minto Creek in 2016 was lower than in 2014 and 2015 but higher than in 2013. Number of taxa at lower Minto Creek in 2016 was within the historical range from 2006 onwards, but moderately lower than in 1994. Lastly, Simpson's Evenness was within the historical range documented after methodological change in 2011. Over the 2012-2016 period benthic invertebrate community metrics parameters were stable, any differences between this time period and earlier sampling was more likely due to a change in mesh size of the Hess sampler, with 2012 used as a bridge.

8.2.3.3 Periphyton

Periphytic algae are simple aquatic plants which inhabit the substrate of water bodies. They can provide a valuable biological monitoring tool to assess potential impacts of nutrient enrichment and metal toxicity. Chlorophyll a is the primary photosynthetic pigment common to all algae. Determining chlorophyll a concentrations provides a measure of algae biomass and thus, the primary productivity of a given location. Periphyton was sampled in 1994 and 2011 to 2016, in Minto Creek (exposure) and Wolverine Creek (reference). Overall, the periphyton community of lower Minto Creek relative to lower Wolverine Creek had lower density and taxon richness. Temporal variability in periphyton community composition was evident among samples taken in 1994 and 2011-2016.

Concentrations of chlorophyll-a have been variable among areas over time. Significant differences between areas only occurred in 2014 and 2015. In 2016, concentrations were higher but as in previous years mean concentrations were below the British Columbia Water Quality Guideline (BCWQG) of 100 mg/m^2 for the protection of aquatic life (BCMOE 1985).

9 VEGETATION

The Minto Mine lies within the Boreal Cordillera ecozone and is situated in the far western part of the Yukon Plateau ecoregion, adjacent to the Klondike Plateau ecoregion in the west. This area was part of the eastern extent of Beringia, which remained ice-free approximately 15–20 thousand years ago. Endemic and rare plant species are associated with the Beringia area as it was a unique and isolated ecosystem. These remnant species are usually associated with grasslands and wetlands.

The Minto property lies within the eastern part of the Dawson Range, with elevations from 700 to 950 m; the landscape has rounded mountains intersected by broad valleys and drainages that are part of the Yukon River watershed. Discontinuous permafrost occurs on northern slopes and low-lying areas where sunlight is reduced.

Forest fires are frequent in this part of Yukon as it lies in the rain shadow of the St. Elias–Coast Mountains and receives less than 300mm of precipitation per year (Smith et al. 2004). As a result, the study area around Minto Mine has experienced numerous fires over the last forty years, rendering it a complex mosaic of plant communities at various stages of succession.

Capstone has characterized the baseline vegetation conditions for the project area by commissioning a vegetation survey and mapping project in 2010 and a vegetation metal uptake program in 2016. The following subsections summarize vegetation surveys conducted in the Project area. For further detail, please see the *Minto Mine Site Characterization – Vegetation* (AEG, 2017) in Appendix 1-7.

9.1 VEGETATION BASELINE SURVEY AND ECOSYSTEM MAPPING

A previous survey was conducted in 1994 by Hallam Knight Piesold (HKP), before any mine development had begun in the area. The 2010 baseline vegetation survey and mapping project was commissioned by Minto to capture the current biophysical conditions prior to expansion activities. The scope of the survey, mapping, and reporting includes classifying forest and vegetation types, identifying sensitive ecosystems, and compiling vegetation and some surficial soils information. In the intervening years between the 1994 and 2010 studies, the local landscape has been altered by the footprint of the current mine, further exploration and three major fires (See Fire History Map, of the *Minto Mine Site Characterization – Vegetation* (AEG, 2017*)*).

In the 2010 Vegetation Baseline Survey, different vegetation communities that exist within the study area were identified through aerial photo interpretation and delineated into polygons. Since most of the study area is regenerating from past fire disturbances, a mosaic of vegetation communities at different successional stages was delineated and classified. Thirty plots (23 of which are permanent plots) were visited to determine the accuracy of photo interpretation.

Most of the study area is regenerative young forest or shrub-dominated ecosystems where tree and shrub species are at a uniform height, as is common in fire disturbed areas (Oswald and Brown 1990). Willows

and trembling aspen are the most represented species in crown cover at the time of the survey. Lodgepole pine is a later successional species and will gradually dominate mid and upper slopes that are well drained. Shade-tolerant white spruce was often found in the understory as seedlings and is a climax species that will eventually overgrow the pine and trembling aspen communities on northern aspects. Black spruce is also a climax species that is adapted to wetter, cooler sites, and is often the persistent species in white/black spruce mixed areas along slope toes and valley bottoms. Small areas of grasslands are scattered along dry crests and steep south facing slopes as these locations do not retain enough moisture to sustain tree growth and are more likely to contain rare or uncommon plants.

Most of the planned expansion is along ridge tops and mid slopes. The main vegetation types in these areas are upland willow species, trembling aspen, lodgepole pine, and associated understory growth. The potentially impacted polygons were surveyed for rare and endemic plants; and while none were found at the time of survey, does not mean that none exist.

The ecosystem map (Appendix 1-7) was designed to be used as a land management and planning tool. As the mine expands its footprint, the map can be referred to for a quick assessment of which type of vegetation communities will be directly disturbed and how much area is involved. Sensitive areas that should be avoided, if possible, include: riparian corridors, bog/wetlands, mature forests, and grasslands.

9.1.1 Vegetation Types

Vegetation types are based on reoccurring patterns of plant associations dictated by site attributes such as: moisture and nutrient availability, aspect, and elevation. Polygons were delineated based on tree/shrub composition, cover, and structural stage. Plot descriptions are more detailed, but most polygons will fit in one of the following general forest types:

Trembling Aspen/Lodgepole Pine

This association is found in early successional forests originating after fire disturbance, on mesic to subxeric sites. Lodgepole pine (*Pinus contorta latifolia*) is more dominant than trembling aspen (*Populus tremuloides)* on well-drained south facing slopes and terraces. Coarse soils are often exposed, and lichens are well represented in the ground cover. Typically, these sites have low growing shrubs such as lingonberry (*Vaccinum vitis idaea*), kinnikinnick (*Arctostaphylos uva-ursi*), and prickly rose (*Rosa acicularis*).

Black Spruce/Labrador Tea/Sphagnum

Found in low lying areas and north facing slopes (cool sites), usually sparse to open forests (<50% crown cover), common shrubs in this ecosystem include Labrador tea, scrub birch, willow, and bog blueberry. Herbs present were sweet coltsfoot (*Petasites frigidus*), cloudberry (*Rubus chamaemorus*), and horsetail (*Equisetum sp.)* Sites are poorly drained (hydric to mesic) with peat horizons over mineral soils, often associated with permafrost.

White/Black Spruce

This association is typically located on south facing lower slopes with upland willow species and Labrador tea. A thick carpet of feather mosses and sphagnum covers the mineral soil. Ground cover shrubs include lingonberry, bog blueberry, and crowberry.

Willow/Trembling Aspen

This was the most common vegetation association in study area, indicative of regenerative growth (>10yrs) after a fire event. Most trees and shrubs are less than 5 m tall; cover can be open to closed as the canopy layer is of uniform height. Other species that may be present include: Alder (*Alnus crispa*) and Alaskan birch (*Betula neoalaskan*) on north facing slopes. Lodgepole pine and white spruce are also present in the understory and will eventually overtop other competing species to form the dominant canopy as the forest matures. The moisture regime ranges from subhygric to subxeric.

Willow/Scrub Birch

Willow (*Salix sp.)* and scrub birch (*Betula glandulosa*) occur in fluvial ecosystems adjacent to streams and fens. Other shrubs present are bog blueberry (*Vaccinum uliginosum*), Labrador tea (*Ledum groenlandicum*), and shrubby cinquefoil (*Potentilla fruticosa*). Associated graminoids include water sedge (*Carex aquatilis*), bluejoint grass (*Calamagrostis canadensis*), and rushes (*Juncus sp*.). Sphagnum, feather, and glow mosses are common.

Trembling Aspen/Grassland

This association features sparse to open cover of trembling aspen (*Populus tremuloides*), often with lodgepole pine (*Pinus contorta latifolia*) present as a minor component. Found on steep south and southwest facing slopes, its understory shrubs include prickly rose (*Rosa acicularis*), soapberry (*Shepherdia canadensis*), kinnikinnick (*Arctostaphylos uva-ursi*), purple reedgrass (*Calamagrostis purpurascens*), Glaucous bluegrass (Poa glauca), Threadleaf sedge (Carex filifolia), Death camas (*Zygadenus elegans*), common yarrow (*Acillea millefolium*), pussytoes (*Antennaria sp*.), and prickly saxifrage (*Saxifraga tricuspidata*).

9.2 VEGETATION METAL UPTAKE

The main objectives of the vegetation metal uptake program were to utilize previously established or documented conditions, monitoring results or predictive efforts, where appropriate and possible; establish a network of plots for monitoring both soil and vegetation metal concentrations; and, allow for an ongoing evaluation of the extent and degree that metals from mining activity is affecting vegetation in proximity to the project site.

These objectives were met for this first round of the program by establishing 16 exposure sites and five control sites that will be used for monitoring every three years. At each site between one and five vegetation samples were collected along with samples of each distinguishable soil horizon.

Results from the 2016 monitoring event present evidence that some metals are elevated in the certain plant species near the mine when compared to control sites. As no baseline vegetation metals samples were collected during the 2009-2010 ecosystem mapping program it cannot be determined that elevated metals were not present in the pre-mining condition.

Results between the rinsed and unrinsed samples were highly variable. Results for some metals were greater for rinsed samples then unrinsed, suggesting that rinsing was not always removing dust particulates. There appeared to be other processes involved since some metals increased after rinsing.

Lichen and willow samples showed significantly higher metals at exposure sites compared to control sites. Lichen species are known to be good indicators of airborne metals and range from being tolerant to sensitive to airborne metals, depending on the physiological form and species (Garty, 2000; Branquinho et al., 2000; Kabata-Pendias et al., 2011). Findings around the Minto site were that lichen samples accumulated more metals than the other plant species and had significantly elevated concentrations of arsenic, cadmium, iron, and lead compared to control sites.

Willows species are known to be accumulators of heavy metals (Kosvakinka and Quigley, 2004). In particular, willows are known to uptake cadmium and zinc from surrounding soils (Tlustos et al., 2007). Willows showed significantly higher arsenic, iron and lead concentrations in exposure than control sites. Conversely and contrary to expectations, willow samples at control sites appeared to have higher (but not significantly higher) concentrations of cadmium and zinc than exposure sites.

Horsetail and Labrador tea appeared to have similar responses, and were both significantly higher in copper and iron at exposure than control sites, but not for arsenic, iron, or lead.

Blueberry did not show any significant difference between control and exposure sites. This may have been due to the short exposure period when the plant is producing berries.

There were some interesting trends in other metals. Soil metals in general were higher at exposure than control sites. Iron concentrations increased significantly for all plant species other than blueberry. Iron is not a COPC, but it does indicate that there were generally higher levels of dust at the exposure than control sites. Zinc concentrations were not higher at exposure than control sites which indicates that zinc was not significantly picked up into plant tissue in the area surrounding the mine. CCME exceedances in soil were limited occuring in two of the 13 metals of interest with 5% of samples for arsenic and 25% of samples for copper exceeding the industrial guideline. When present, copper exceedances were often detected for the depth of the pit which reflects the mineralization of the area.

Large amounts of variation (noted by high standard deviations) are present within the data that were collected in 2016. The sources of variation are potentially attributed to natural variability in soil and plant

tissue, plant species, source types and distance to exposure sample sites, field sampling conditions, and small sample size.

A main factor attributing to field variability for this sampling event was weather. The rain during the beginning of the sampling period could have affected the quantity of metals that were present on species that were more exposed compared to species in closed canopy conditions. There were scattered thunder showers for the remainder of the sampling period that could have affected parts of the study area. This variability is likely to persist through future sampling events as weather is unpredicatble and cannot always be avoided.

Variability attributed to sample size is reflected in the limited number of control sites and the number of target species present within each site. Blueberry sample size was limited for both control and exposure sites as they are less abundant across the landscape compared to the other species.

Laboratory variability was encounted by the nature of the selected lab's data quality objective (DQO). The DQO implemented by the lab for tissue analysis was 40% repeatability. This threshold allows for 40% variation in concentrations between any duplicate samples. This threshold was set by that lab and is a factor of the natural variability that is present in vegetation tissue sampling. This variation is compounded when the lab homogonized, subsampled, and rinsed half of the samples for the purpose of quantifying external and internal metal concentrations. Large variation made comparisons between rinsed and unrinsed samples challanging when there were low metal concentations and minimal differences between subsamples. Additionaly, many of the results received were below the method detection limits. When this was the case, the value used to calculate statistics was half the detection limits. Including these halved detection limits is likely to have increased the variation and standard deviation for each variable.

Vegetation metal uptake surveys will continue on a three-year basis as a condition of the current licence to monitor trends and changes in metal concentrations in vegetation and soil.

10 GEOCHEMISTRY AND GEOTECHNICAL INFORMATION

10.1 GEOCHEMISTRY

Minto conducted pre-production testing and continues to conduct operational monitoring. Detailed information on geochemistry can be found in Appendix 1-8 to this report.

The primary conclusions from the pre-production testing were that Main Pit and Minto North Pit waste rock was expected to be net acid neutralizing, and that the main acid neutralizing mineral was calcite (calcium carbonate, $CaCO_3$). The primary conclusions of preproduction testing for Area 2/ Area 118, Ridgetop, and proposed underground developments were that bulk granodiorite waste rock was expected to be net acid neutralizing, and that some mineralized waste rock was expected to have NP:AP values less than the threshold value of 3 specified by the water use license (YWB, 2015 and its predecessors).

Since mining commenced at Minto, operational ABA monitoring has been carried out on composite samples of blast hole cuttings from each waste blast. Waste rock monitoring reports have been prepared and submitted to regulators as part of the mine's compliance reporting (initially semi-annually, now annually). The results show that Main Pit and Minto North Pit waste rock generally had low AP and NP:AP ratios greater than the threshold value of 3. Results for Area 2/ Area 118, Ridgetop, and proposed underground developments waste rock were mixed with most samples having NP:AP values greater than 3. There were sufficient samples with NP:AP less than 3 that Minto modified its waste rock management strategy to include segregation of NP:AP<3 materials. See Appendix 1-8 and Figure 10-1 for summary results of operational waste rock geochemistry monitoring.

No pre-production characterization tests were carried out on tailings from Main Pit ore. For the Phase IV and Phase V/VI deposits (Area 2, Area 118, Minto North, Ridgetop and Minto South underground), a range of preproduction tests were carried out on residues from locked cycle testing. Residues from metallurgical testing carried out on Area 2 and Area 118 ore samples were subjected to additional testing for purposes of ML/ARD characterization. Results of the pre-production testing on Phase IV and Phase V/VI tailings indicated that the material would be net acid neutralizing, and that risks of neutral pH metal leaching would be similar to those for Main Pit tailings produced from full-scale operations.

Operational monitoring of as-produced tailings has been carried out to track the ABA characteristics of the tailings produced over the operating period. Results of operational ABA monitoring of tailings produced to date confirm excess of neutralization potential (NP) over acid potential (AP), indicating that the tailings produced to date are net acid neutralizing. See Appendix 1-8 and Figure 10-2 for summary results of operational tailings geochemistry monitoring.

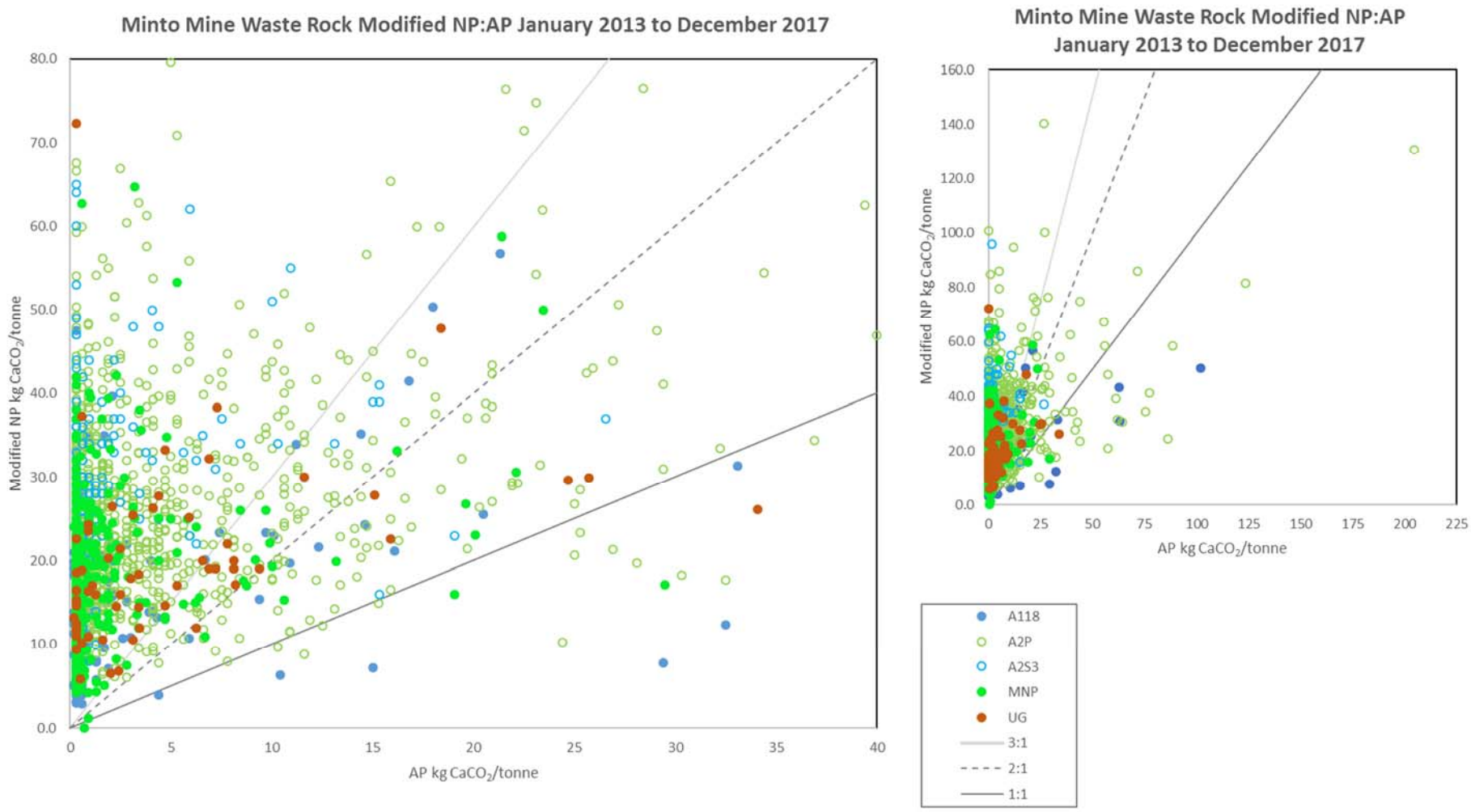


Figure 10-1: Minto Mine Waste Rock Modified NP:AP January 2013 to December 2017.

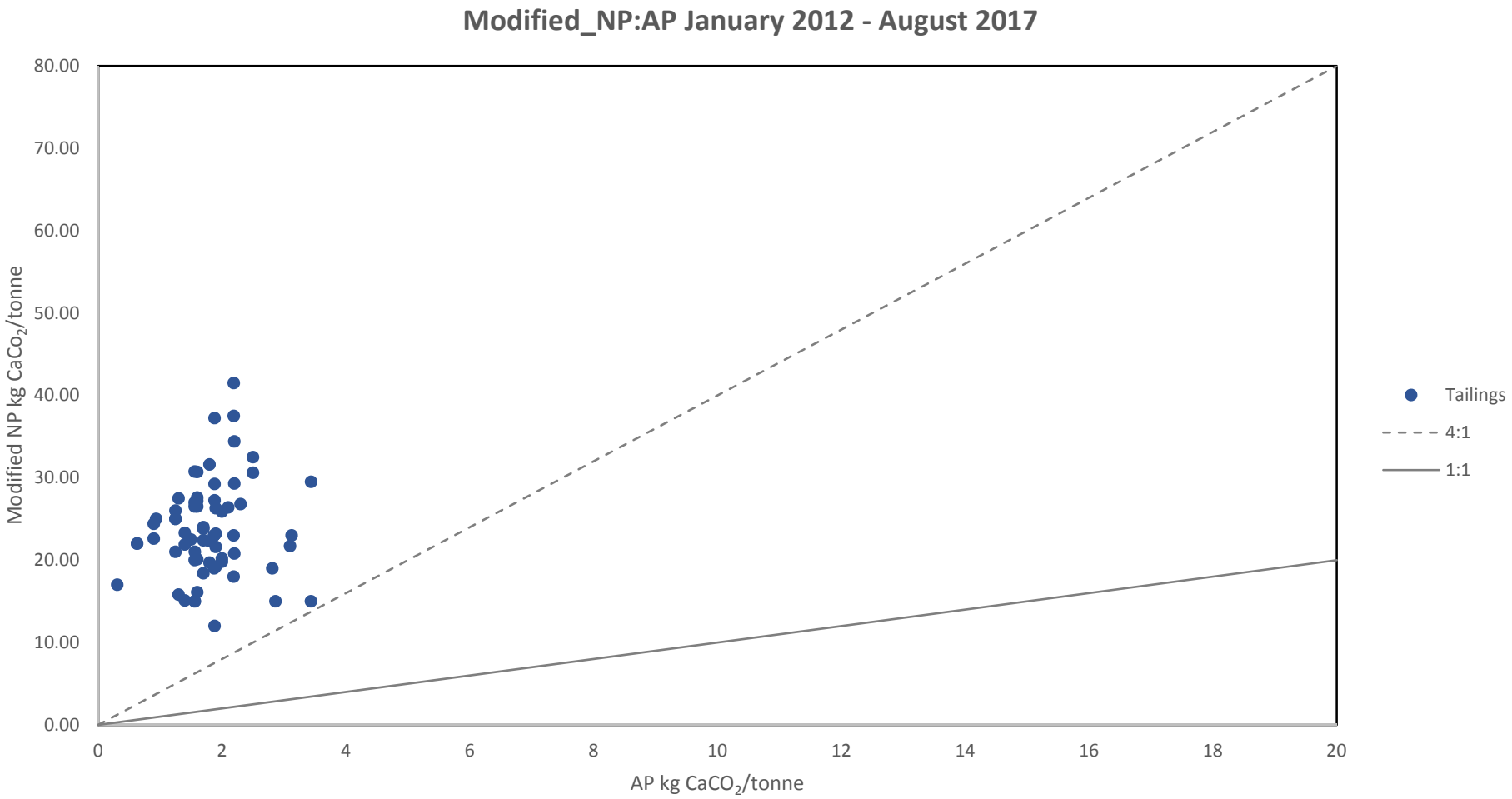


Figure 10-2: Modified NP:AP January 2012 – August 2017.

10.2 GEOTECHNICAL INFORMATION

The geotechnical stability of individual structures is addressed in geotechnical annual review reports. The site has in the past experienced slope instabilities and slow movements along a suspected shear zone underlying parts of some existing facilities; the physical monitoring program is carried out to monitor geotechnical performance of major site facilities, and review of monitoring results is an important component of the geotechnical annual inspections. The report documenting the most recent geotechnical annual review (SRK 2017) is attached as Appendix 1-9.

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Prepared for

Minto Explorations Ltd.



Prepared by

SRK Consulting (Canada) Inc.
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February 2018

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Appendices

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Appendix B: Water and Load Balance Source Terms
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1 Introduction

1.1 Purpose and Overview

This report describes the development and results of the water and load balance model developed for the Minto Mine 2018 Reclamation and Closure Plan (RCP). This report was written in accordance with the Water Use License (WUL) requirements (YWB 2015; Clause 110c). The main features of the 2018 RCP are summarized in Section 1.2.

The water and load balance model builds on previous water balance modelling work prepared for Minto Explorations Ltd. as part of past water license applications, environmental assessments, annual reporting and operational water management efforts. The model covers the end of the operations phase, as well as the active closure and post-closure phases (Post-Closure I and Post Closure II).

The water balance model accounts for all inflows to and outflows from the Minto Mine site, as well as accumulation of water in reservoirs within the mine site. The model can be run in stochastic mode where annual precipitation rates are varied to simulate randomly occurring wet and dry climatic conditions. Water quality results are based on estimates of geochemical loadings applied to modelled water flow. Water quality predictions are presented for 'expected case' and 'reasonable worst case' scenarios, which represent the best estimate geochemical loading rates and loading rates that are unlikely to be exceeded.

Section 2 describes inputs to and components of the water balance model. Section 3 summarizes the load balance model inputs. Model scenario implementation is described in Section 4 and modelling results are presented in Section 5 followed by a discussion in Section 6.

1.2 Closure Summary

Closure at the Minto Mine has been subdivided into three periods: Active Closure (AC), Post-Closure I (PCI) and Post Closure II (PCII) (Minto 2018). Estimated timelines for the different phases are listed in Table 1-1.

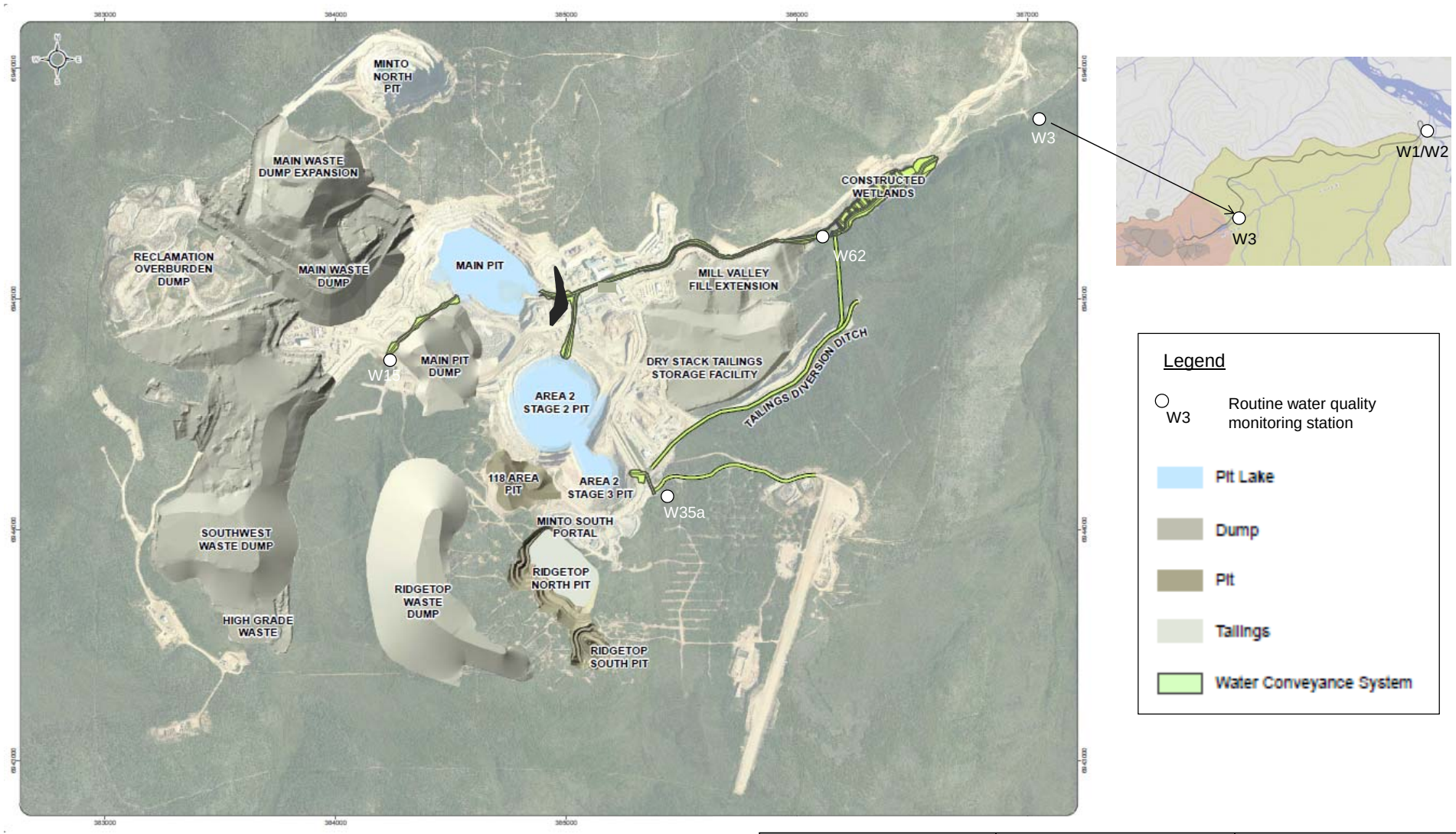
Table 1-1 Project Schedule

Year	Summary of Main Project Activity
2018	Underground production continues from the Minto South underground portal. Open pit mining continues in Area 2 Stage 3 until the pit is completed by the end of the first quarter of the year. Open pit mining begins at Ridgetop. Underground mining begins at the Copper Keel deposit
2019	Open pit mining continues at Ridgetop until the pit is complete in the second quarter. Underground mining continues at the Minto East deposit until it is completed in the first quarter. Underground mining begins at the Minto East 2 deposit.
2020	Underground mining begins at the Minto North Underground deposit.
2021	Underground mining from the Minto East and the Minto North Underground deposits and milling continues until the last quarter of the year. Active closure to begin.
2022 – 2024	Active closure period continues (total duration expected three years) with post closure one beginning near the end of 2025.
2025 – 2029	Post closure one period continues (total duration expected five years but will be performance based) with post closure two beginning near the end of 2030
2030 – 2039	Post closure two period continues (total duration of post closure two is ten years) until near the end of 2040.

The AC period includes the implementation and construction of most closure measures, including recontouring, soil placement, closure water conveyance construction, revegetation, construction of passive water treatment facilities, decommissioning of the Water Storage Dam, demobilization and demolition. During the AC period, operational water quality objectives/effluent standards will still apply. Active treatment will be used as required to meet discharge water quality limits. The AC period is expected to span approximately three years.

In the PCI period, the closure measures and infrastructure will become operational and the performance will be assessed. During this period, maintenance on soil covers will be undertaken as required as vegetation begins to establish. Water conveyance features will be commissioned along with the passive treatment system. During PCI, operational water quality objectives/effluent standards will still apply. A pump-back system will be established downstream of the passive treatment system to direct any water that does not meet effluent standards to the treatment plant or the open pits. Passive treatment will be commissioned and will over time become the primary water treatment system. The PCI phase nominally spans five years but the duration of this phase will depend on the performance of the closure measures.

During the PCII period, the closure measures and infrastructure will be monitored to confirm that performance meets the design intents. PCII is primarily a monitoring phase with maintenance activities as required. During PCII the closure water quality objectives will apply and passive treatment will be the primary water treatment onsite.



Source: Adapted from Figure 7-2 (prepared by Access Consulting Group) in Minto Explorations Ltd. 2018. Minto Mine Reclamation and Closure Plan Revision 2018-01.

srk consulting	capstone MINTO MINE	Water and Load Balance Model Report 2018
		General Arrangement- End of Mine
Job No: 1CM002.044 Filename: Fig_1-1_EOM_GeneralArrangement.pptx	Minto Mine	Date: Feb 2018 / Approved: DBM / Figure: **1-1**

2 Water Balance Model Description

2.1 Water Balance Overview

Figure 2-1 shows a schematic of the conceptual water balance for the Minto Site. The water balance can be described as:

$$Water\ Input = Water\ Storage + Water\ Output \quad (1)$$

Precipitation is the only source of water to site and therefore the only input. The open pits and the Water Storage Pond (WSP) are the primary water storage reservoirs on site during the Operational Period. Water outputs include water released to Minto Creek and water lost to evapotranspiration. The net input of water to reservoirs on site, or net yield, can be expressed as:

$$Net\ Yield = Precipitation + Groundwater\ Discharge\ - Evapotranspiration - Groundwater\ Recharge \quad (2)$$

Note that the yield term includes both surface and groundwater terms. Net yield is a measure of the total volume of water that report to the reservoirs on site, regardless of flow path. Net yield can be described in terms of a yield coefficient, defined as follows:

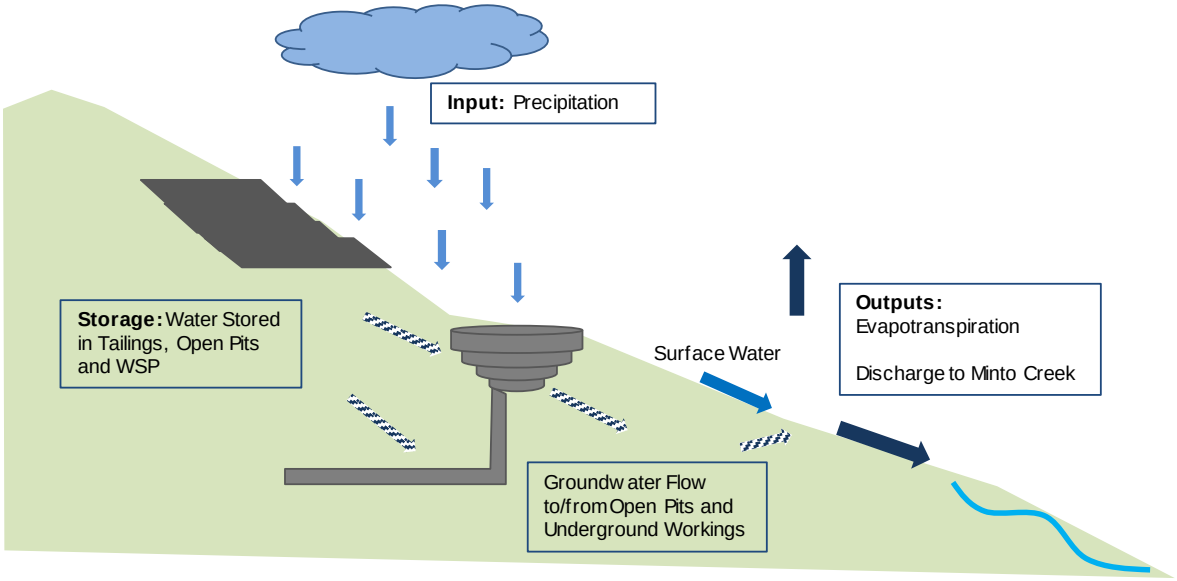
$$Net\ Yield = Precipitation * Yield\ Coefficient \quad (3)$$

The yield coefficient approach allows the water balance to be simplified to:

$$Net\ Yield - Water\ stored\ on\ Site = Water\ Released\ to\ Minto\ Creek \quad (4)$$

Figure 2-1 shows a schematic of the simplified water balance. In a year with average annual precipitation (329 mm/year) the net yield from catchments within the Minto Mine site is approximately 950,000 m^3. Of that, approximately 530,000 m^3 will report to tailings pores during the period when the mill is running. The annual volume of water reporting to waste rock pores range from zero to about 200,000 m^3, depending on the volume of waste inundated in any given year. On average, waste rock pore water amounts to roughly 60,000 m^3/year. This leaves approximately 360,000 m^3 of surplus water that must be stored on site or released to Minto Creek in an average-precipitation year.

In previous versions of the water and load balance model for the Minto Mine, groundwater was not included as a discrete path. At Minto, the cross section of the Minto Creek Valley at the Water Storage Dam is narrow and bedrock is relatively shallow. As such, the majority of groundwater that flows toward Minto Creek is expected to report to the stream as surface water and only a small fraction is expected to flow from site via subsurface pathways (SRK, 2015a). In this 2018 revision of the water and load balance model, the assumption that the majority of the groundwater reports to the surface upstream of the Water Storage Pond (WSP) is carried forward. However, discrete groundwater discharge and recharge terms for the open pit zones have been included. The groundwater flow estimates are based on a revised version of the 2015 Groundwater Model as described in SRK (2015a).



Source: X:\01_SITES\Minto\1CM002.0XX_PhaseVII_Support\!080_Deliverables\Water and Load Balance Report\040_Figures\Figure 2.1 and 2.2 Water_Balance_Schematic_Minto_REV01.pptx

Figure 2-1 Water Balance Schematic

2.2 Water Balance Inputs

Inputs for the water balance model are summarized in Table 2-1 and discussed in the following sections.

Table 2-1 Input Required for Water Balance Model

Water Balance Component	Input Required
Net Yield	Annual precipitation rates Open water evaporation rates Sub-catchment areas Site-wide yield coefficient Typical hydrograph
Water storage	Open pit water volumes Tailings and waste rock deposition schedules
Groundwater	Estimated seepage rates to/from open pits Underground dewatering rates
Water Discharged to Minto Creek	Historical WSP pumping schedule

2.2.1 Precipitation

Precipitation is the only input of water to the Minto Mine site water balance model. The water balance model calculates annual yield based on annual precipitation and a site-wide yield coefficient. Therefore, a reliable estimate of the frequency and intensity of total annual precipitation is important for the accuracy of model predictions.

Rainfall data collected on site was used in conjunction with long-term regional precipitation records to estimate the distribution of annual precipitation rates. The precipitation record for the Minto site is shown in Table 2-2. Rainfall data has been recorded at site beginning in September 2005. In October 2011, Minto installed an adapter for measuring snowfall as water equivalent. For this update, on-site rainfall and total precipitation data from the period September 2005 through December 2016 were used.

Table 2-2 Monthly Rainfall and Total Precipitation Measured at Minto

Year	Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	Total
2005	**	**	**	**	**	**	**	**	21.60	23.00	11.60	4.60	60.8
2006	0.00	0.40	2.80	**	22.84	35.80	28.63	29.20	12.20	12.20	0.00	0.00	144.1
2007	0.20	0.00	0.40	5.80	4.60	36.00	47.80	21.00	33.80	11.80	0.00	0.00	161.4
2008	1.20	2.00	0.80	1.80	9.60	26.20	**	100.60	21.80	6.40	0.00	0.00	170.4
2009	5.20	0.00	0.80	3.23	**	**	6.08	50.76	7.20	16.60	0.00	0.00	89.9
2010	0.00	0.00	0.00	0.00	7.60	48.8	75.6	46.4	18.0	16.3	3.05	0.00	215.8
2011	0.00	6.40	0.20	0.40	15.3	56.0	101.8	64.8	15.6	4.40(A)	0.15**	3.94**	269.0
2012	9.0**	9.9**	34.9	0.0	0.1**	32.1	44.8	20.6	26.1	16.5	17.1	18.4	229.5
2013	4.4	73.8	7.4	0.0	7.9	21.0	113.5	46.8	59.7	13.6	36.6	27.0	441.7
2014	16.9	8.0	0.0	3.8	15.0	12.0	50.5	13.4	30.5	22.0	2.9	21.0	196.0
2015	12.4	0.0	10.9	8.0	4.9	20.5	37.7	80.3	16.7	27.7	7.1	11.5	237.7
2016	11.7	7.8	3.3	14.7	31.5	22.2	68.2	35.9	17.8	9.8	13.6	14.1	250.6

Source: Minto Site Data: X:\01_SITES\Minto\1CM002.024_Water_Balance_Support\2015_Water_Balance_Update\Minto Water Balance\2016 Met Station DataSummary.xlsx

Notes:

** partial data only.

(A) Measurement transitioned from rainfall to total precipitation on Oct. 15/2011.

Green highlight: total precipitation measurements.

Regional precipitation data were available from a number of meteorological stations in the Yukon (Table 2-3). Rainfall data up until 2012 from each station were correlated with precipitation data collected from the Minto site to determine whether long-term regional precipitation data would be suitable for use as a basis for estimating annual precipitation at the site.

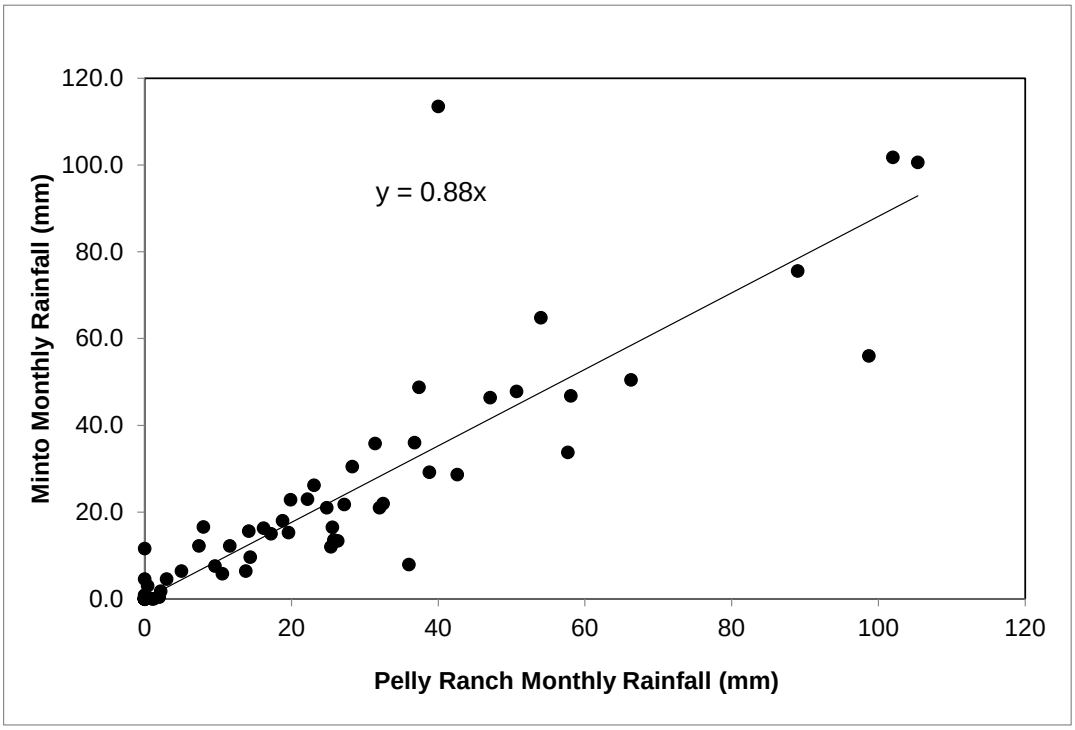
Rainfall data at the closest regional meteorological station at Pelly Ranch (Climate ID: 2100880), located 23 km from the Minto Mine, resulted in the best correlation with rainfall data from site. The precipitation record at Pelly Ranch dates back to December 1951 and has continued to the present day. Figure 2-2 shows a plot of monthly rainfall values at Minto and at the Pelly Ranch station. The correlation can be expressed as:

$$Minto\ Rainfall = 0.88 * Pelly\ Ranch\ Rainfall \quad (6)$$

Table 2-3 Regional Meteorological Stations

Station	Latitude	Longitude	Distance From Site (km)	Elevation (m)	Record Begins	Record Ends
Minto Site	62°36'59"	137°15'00"	0	887	Sep-05	Present
Burwash Airport	61°22'00"	139°03'00"	168	807	Oct-66	Present
Carmacks	62°06'00"	136°18'00"	109	525	Aug-63	Present
Dawson Airport	64°02'35"	139°07'40"	184	370	Jan-76	Present
Faro Airport	62°12'27"	133°22'33"	205	717	Dec-77	Present
Haines Junction	60°46'21"	137°34'49"	206	595	Oct-44	Present
Mayo Airport	63°37'00"	135°52'00"	131	504	Oct-11	Present
McQuesten	63°36'00"	137°31'00"	110	457	Oct-86	Present
Pelly Ranch	62°49'00"	137°22'00"	23	454	Dec-51	Present
Whitehorse Airport	60°42'34"	135°04'07"	241	706	Apr-42	Present

Source: SRK, :\01_SITES\Minto\!020_Site_Wide_Data\Water_and_Load_Balance_Files\02_Hydrology_and_Meteorology\Regional_Analysis\Regional Precipitation Data\Regional_Met_Station_Summary_1CM002.003_Rev01_TC.xlsx



Source: X:\01_SITES\Minto\1CM002.024_Water_Balance_Support\!080_Deliverables\2016_RCP_Water_and_Quality_Report\040_Figures\Figure 2-3.xlsx

Figure 2-2 Comparing Minto Mine and Pelly Ranch Rainfall (Sept. 2005 to March 2015)

A comparison of snow-water equivalent data from Minto and Pelly Ranch (not shown) resulted in the following correlation:

$$Minto\ Snowfall = 1.24 * Pelly\ Ranch\ Snowfall \quad (7)$$

Based on these correlations the annual precipitation at Minto correlates to the annual precipitation at Pelly Ranch as follows:

$$Minto\ Annual\ Total\ Precip. = 1.09 * Pelly\ Ranch\ Annual\ Total\ Precip. \quad (8)$$

Table 2-4 shows the estimated average total annual precipitation values for Minto based on the correlation with Pelly Ranch precipitation data. A frequency (or probability) distribution of total annual precipitation was developed for the Minto site based on the long-term total precipitation record available for the Pelly Ranch station. Pelly Ranch precipitation values were only used if a calendar year contained valid information for more than 95% of the total number of days in that year. The use of this data quality criteria resulted in a total of 51 years of valid data to be included in the frequency distribution analysis.

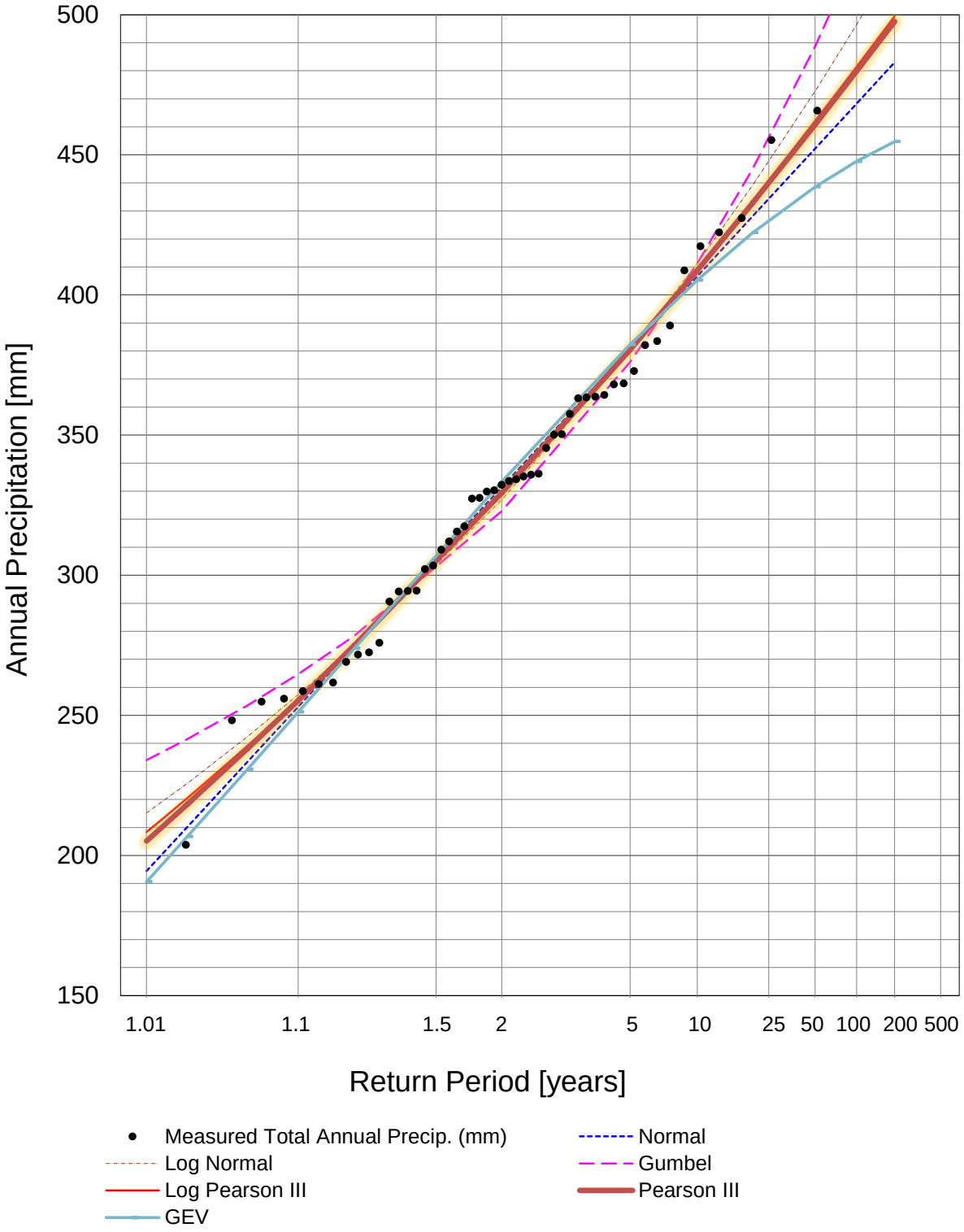
The 51 years of annual precipitation data were fitted against six different statistical distributions: Normal, Log-Normal, Gumbel, Log-Pearson, Pearson III, and GEV. The best fit was obtained with a Pearson III Distribution ($r^2 = 0.99$). This Intensity/frequency distribution was used as input for the water balance model. The final results are presented in Figure 2-3 and Table 2-5.

Table 2-4 Estimated Average Precipitation at Minto Based On Pelly Ranch Precipitation Data

Month	Average Rainfall (mm/month)	Average Snowfall (mm/month)	Average Total Precipitation (mm/month)
Jan	0.08	28.6	28.7
Feb	0.08	20.3	20.4
Mar	0.14	15.9	16.0
Apr	3.10	9.29	12.4
May	19.16	0.63	19.8
Jun	32.38	0.00	32.4
Jul	47.85	0.00	47.9
Aug	34.98	0.02	35.0
Sep	22.91	3.07	26.0
Oct	6.72	22.0	28.7
Nov	0.30	37.3	37.6
Dec	0.08	30.4	30.4
Total (mm/year)	167.8	167.5	335.3

Source:Z:\01_SITES\Minto\!020_Site_Wide_Data\Water_and_Load_Balance_Files\2_Hydrology_and_Meteorology\Met_Data\Minto_MasterStationFile_20130607.xlsx

The estimated average total precipitation for Minto was slightly lower for the frequency analysis (329 mm/year) than for the direct computation using the average total precipitation value from Pelly Ranch (335 mm/year). This difference is a result of the imperfect fit between the data and the distribution model that is used. In the interest of consistency with frequency distribution approach, the total precipitation value generated by the Pearson III distribution (329 mm) was used as input to the water balance model.



Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\PMP & PMF\PMP Hershfield Estimation & Freq Distribution_VM_Rev4_SRJ.xlsx

Figure 2-3 Precipitation Frequency Analysis for the Minto Mine Site

Table 2-5 Precipitation Frequency Analysis Statistics

Cumulative Probability	Return Period	Season	Normal	Log Normal	Gumbel	Log Pearson III	Pearson III	GEV
0.005	200		483	519	554	499	498	455
0.01	100		468	496	521	481	480	448
0.02	50		452	473	489	462	461	439
0.05	20	Wet	428	439	445	434	433	422
0.1	10		407	411	411	409	409	405
0.2	5		381	380	376	380	380	382
0.5	2		331	327	323	329	329	333
0.8	5		275	275	278	275	275	274
0.9	10		253	257	265	256	255	251
0.95	20	Dry	233	242	254	239	238	231
0.98	50		210	226	242	220	218	207
0.99	100		194	215	234	208	205	191
	R^2		0.98	0.98	0.97	0.98	0.99	0.97

Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\PMP Hershfield Estimation & Freq Distribution_VM_Rev4_SRJ.xlsx

2.2.2 Evaporation

For modelling purposes evaporation estimates were based on regional data. Monthly lake evaporation (aka potential evaporation) has been recorded at the Pelly Ranch station from 1965 to 2005; the mean annual lake evaporation value is 452 mm. In historical revisions of Minto's water balance, open water evaporation and evapotranspiration were estimated based on the regional data and model estimates.

Estimated evaporation values for the site were adopted from work completed by Clearwater Consultants Ltd. as part of the previous year's water balance update completed for the mine site (CCL 2010). Table 2-6 shows the adopted monthly evaporation values, which for modelling purposes were assumed to be constant for each year included in the scenario runs.

Evaporation estimates are associated with considerable uncertainty. However, in the present model revision, evaporation losses were only discretely applied to open water bodies, including the pit lakes and the Water Storage Pond. A sensitivity analysis was completed to quantify the effect of uncertainties associated with the annual evaporation rate. The analysis showed that a difference of +/- 100 mm of annual evaporation would make a +/- 2.6% difference to the annual water balance (net inflow) estimate, which was considered to be relatively insignificant in the context of other uncertainties and year-to-year variability.

Table 2-6 Estimated Monthly Open Water Evaporation Values for Minto

Month	April	May	June	July	August	September	Total
Evaporation (mm)	12	83	119	112	83	30	439

Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\Water_and_Load_Balance_Files\01_Project_Phases\04_Amendment_8_Support\02_Source_Term_Archive\[Minto Mine Water Balance_2011 Update Modified Goldsim_SRJ_Rev01

2.2.3 Catchments

The Minto Mine site, excluding the proposed Minto North Pit, is located within the Upper Minto Creek watershed. For the purposes of this report, Upper Minto Creek will refer to the portion of the Minto Creek catchment upstream of the Water Storage Dam. The catchment downstream of the Water Storage Dam will be referred to as Lower Minto Creek. The Upper Minto Creek catchment covers an area of 1,038 ha, which has been divided into sub-catchments for modeling purposes. Table 2-7 and Figure 2-4 show Upper Minto Creek sub-catchments that were delineated for the Phase V/VI water balance model.

The Minto North Pit is located within the McGinty Creek catchment to the north of the Minto Mine. Table 2-8 and Figure 2-5 shows McGinty Creek sub-catchments delineated for the Phase V/VI water and load balance model.

Table 2-7 Upper Minto Creek Catchment Areas

n1_Main_Pit_Undisturbed	**11.5**
n2_SW_Waste_Rock	**40.7**
n3_Overburden_Dump	**31.5**
n4_Overburden_Dump_Undisturbed	**16.6**
n5_Main_Waste_Dump_Undisturbed	**12.8**
n6_Main_Waste_Dump	**48.1**
n7_SW_Waste_Rock_Undisturbed	**47.6**
n8_W15_Undisturbed	**110.9**
n9_Main_Pit_Highwall	**26.3**
n10_Mill_Area_Undisturbed	**27.6**
n11_Area_2_Pit	**18.3**
n12_Portal_Area	**2.2**
n13_Ridgetop_N_Highwall	**2.9**
n14_Mill_Area	**16.0**
n15_Ore_Stockpiles	**13.3**
n16_WSP_Undisturbed_N	**139.8**
n17_Dry_Stack	**19.3**
n18_Dry_Stack_Shell	**5.6**
n19_Main_Pit_Submerged	**15.2**
n20_Ridgetop_N_Pit	**8.2**
n21_Mill_Valley_Fill_Stage_1	**12.4**
n22_Ridgetop_Waste_Dump	**28.1**
n23_Area_118_Pit	**6.9**
n24_Ridgetop_South_Pit	**2.3**
n25_Main_Pit_Dump	**14.9**
n26_Portal_Undisturbed	**7.6**
n27_Area_118_Undisturbed	**2.7**
n28_Ridgetop_N_Undisturbed	**13.1**
n29_W35_Undisturbed_South	**144.1**
n30_WSP_Undisturbed_S	**66.5**
n31_Area_2_Highwall	**17.7**
n32_Mill_Valley_Fill_Stage_2	**4.7**
n33_Underground_Laydown	**2.9**
n34_Drystack_Undisturbed	**4.8**
n35_Tails_Diversio_Upstr_Undis	**27.9**
n36_W35_Undisturbed_East	**28.8**
n42_Contractor_Area	**5.0**
n43_Overburden_undisturbed	**1.1**
n44_medium_grade_waste	**22.2**
n45_high_grade_waste	**5.8**
n46_ice_rich_overburden	**4.2**
SUM	**1038**

Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\Water_and_Load_Balance_Files\04_Catchments\RCP_Catchments\Minto_Phase_VII_RCP_Catchment_Table



N

0 100 200 400 600 Meters

Legend

Subcatchments

srk consulting

Job No: 1CM002.58

Filename: Minto_20171101_SubCatchments_rev00

capstone MINTO MINE OPERATED BY MINTO EXPLORATIONS LTD.

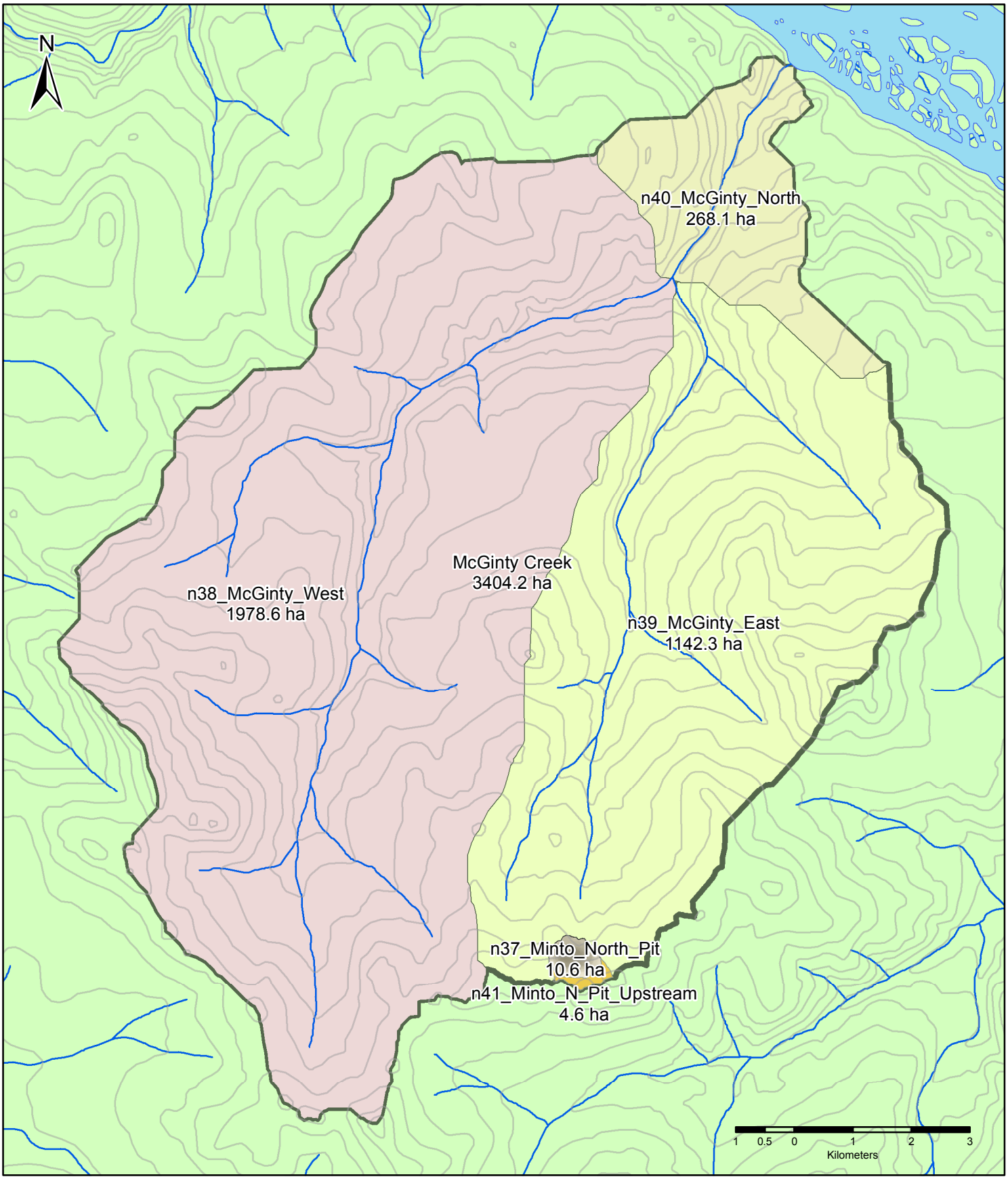
Minto Mine

Water and Load Balance Report
Report 2018

Subcatchments within
Minto Mine Site

Date:	Approved:	Figure:
Feb 2018	SRJ	2.4

Document Path: X:\01_SITES\Minto\!020_Site_Wide_Data\!GIS\0.figures\Minto_20171101_SubCatchments_rev00.mxd



srk consulting	CAPSTONE MINING CORP. MINTO MINE OPERATED BY MINTO EXPLORATIONS LTD.	Water and Load Balance Report 2018
		McGinty Creek Sub-Catchments
Job No: 1CM002.024 Filename: PhV_VI_Minto_North_REV06_AMD	Minto Mine	Date: February 2018 | Approved: KNK | Figure: 2.5

Table 2-8 McGinty Creek Sub-Catchments

Sub-Catchment	Area (ha)
n37_Minto_North_pit	10.6
n38_McGinty_West	1978.6
n39_McGinty_East	1142.3
n40_McGinty_North	268.1
n41_Minto_N_Pit_Upstream	4.6
Total Catchment Area	**3404.2**

Source:X:\01_SITES\Minto\!020_Site_Wide_Data\Water_and_Load_Balance_Files\04_Catchments\Phase_V_VI_Catchments\Minto_Phase_V_VI_Catchment_Table.xlsx

2.2.4 Yield Coefficient

As described in Section 2.1, annual catchment yield (water that reports to reservoirs on site) can be estimated by multiplying the total annual precipitation by a site-wide yield coefficient (Equation 3). This yield includes surface runoff and steady-state flow of shallow and deep groundwater. Groundwater flows to or from the open pits or underground workings are not included in the yield estimate. Such groundwater flows are caused by advancement and dewatering of underground workings or by inundation of open pits and are therefore not influenced directly by the same-year precipitation and run-off. For the 2018 water and load balance, the use of a site-wide yield coefficient for land areas was found to be more appropriate than assigning specific yield coefficients to areas with different land use and surface cover characteristics. Firstly, flow measurements on site (hydrological monitoring stations and flow meters) measure water collected from different land use areas, *i.e.* a combination of undisturbed and developed mine areas. This makes it difficult to evaluate yield coefficients for any one type of land use area based on actual site performance data. Therefore, uncertainties associated with area-specific yield coefficients would magnify the uncertainty of the site-wide water balance and it would be necessary to arbitrarily adjust each of the yield coefficients to match the site-wide water balance. Secondly, during the operations and closure (pit filling) phases the available water storage capacity on site is generally greater than the annual volume of surface yield, and therefore reliable forecasts of total site-wide annual yield volumes are more important for water management planning than forecasts of yield from individual sub-catchments.

Monitoring data collected on site since 2007 was used to estimate the value of the site-wide yield coefficient for Upper Minto Creek. The yield coefficient estimate is updated once a year in conjunction with the annual water balance update for the site. The 2015 and 2016 water balance updates (Appendix B) generally agreed with the established yield coefficient value of 0.30, which was derived based on previous years' water balance data.

The estimated total annual precipitation at the Minto Mine has generally been greater than 300 mm/year for the period 2007 to 2015. Consequently, the site-wide yield coefficient of 0.30 was effectively derived for annual precipitation greater than 300 mm/year. However, this yield coefficient may overestimate yield in relatively dry years. In order to account for this the site-wide yield coefficient for dry years was assigned lower values in the water balance model. The dry year coefficients were based on work completed by Clearwater Consultants Ltd. (CCL 2010) for the Minto Mine site as follows:

- A yield coefficient of 0.15 was used for dry years with less than 190 mm total precipitation.
- A yield coefficient of 0.30 was used for years with greater than 309 mm total precipitation.
- Yield coefficients for years with total precipitation between 190 mm and 309 mm were interpolated values between 0.15 and 0.30.

Figure 2-6 shows yield coefficient values used in the model as a function of total annual precipitation.

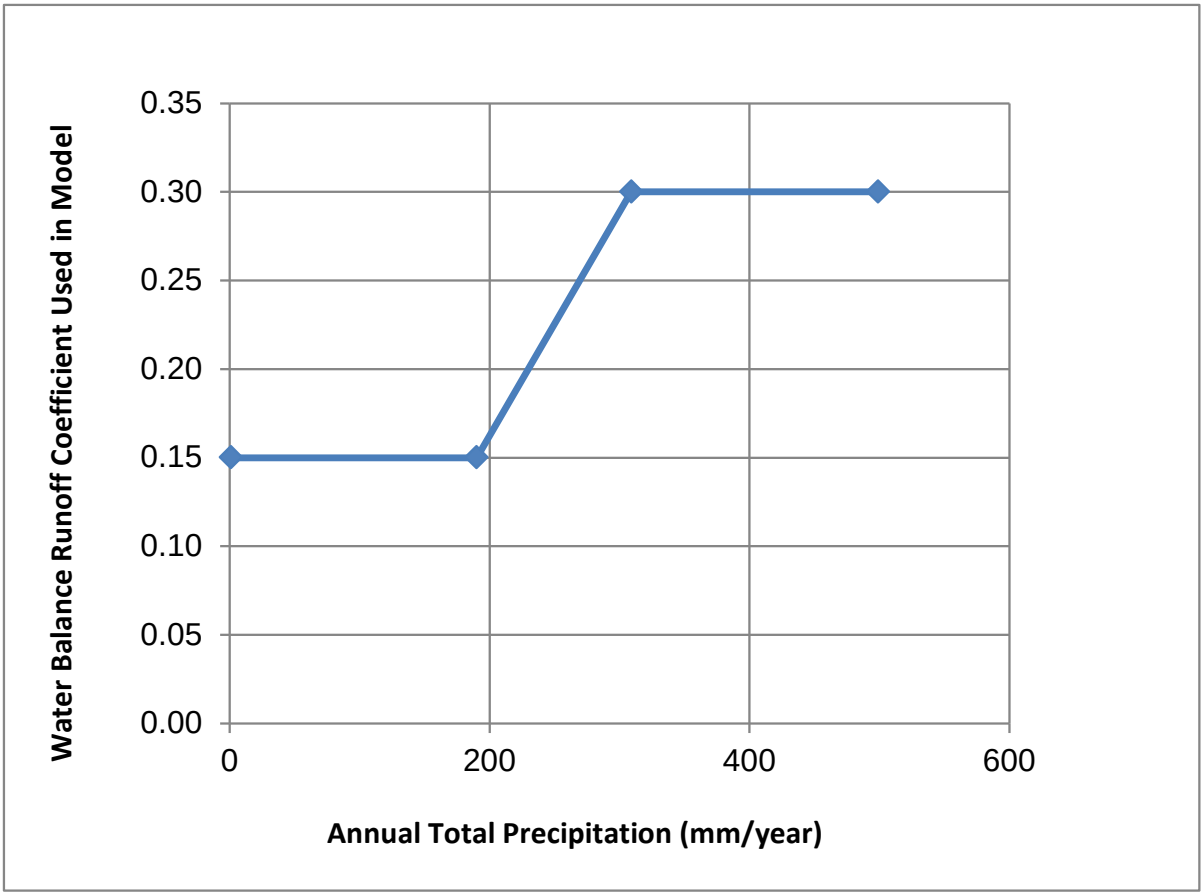


Figure 2-6 Water Balance Yield Coefficient vs. Total Annual Precipitation

The one exception is that open water areas (flooded pits and the WSP) were treated differently than land areas. Open water was assigned a yield coefficient of 1.0 along with an annual evaporation rate. This approach relies on the assumptions that any precipitation that falls on a water body becomes part of the water inventory and that the open water evaporation rate is relatively constant from year to year. As discussed in Section 2.2.2, the effect of uncertainty associated with the estimate of annual evaporation on the site-wide water balance is relatively insignificant.

Catchment yield estimates calculated from the annual precipitation frequency distribution, catchment area and site-wide yield coefficient are listed in Table 2-9.

Table 2-9 Annual Site-Wide Runoff Estimates

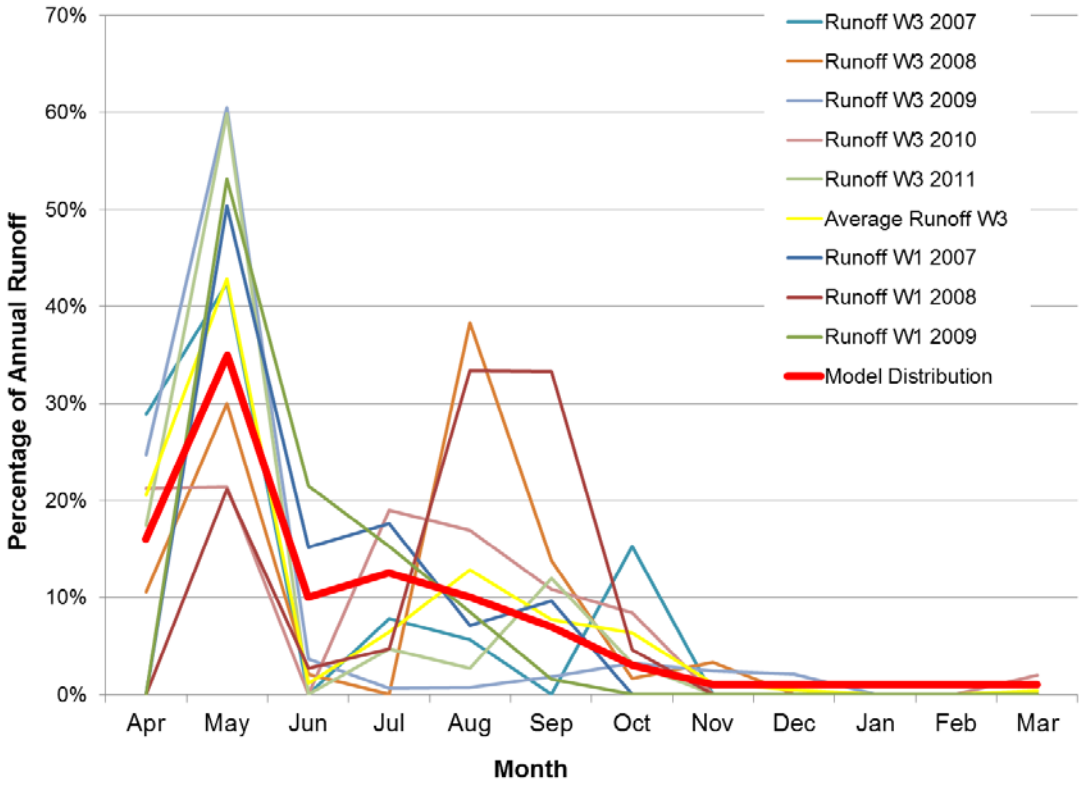
Annual Total Precipitation	Description	Total Catchment Runoff
mm/year		m^3/year
498	1:200 Wet	1,480,000
480	1:100 Wet	1,420,000
461	1:50 Wet	1,360,000
433	1:20 Wet	1,280,000
409	1:10 Wet	1,200,000
380	1:5 Wet	1,110,000
329	**Average**	**950,000**
275	1:5 Dry	660,000
255	1:10 Dry	540,000
238	1:20 Dry	450,000
218	1:50 Dry	350,000
205	1:100 Dry	290,000

Notes: Source:\ X:\01_SITES\Minto\1CM002.024_Water_Balance_Support\2016_Water_Balance_Update\ 2016 Water Balance Update REV00 KNK.xlsx

2.2.5 Runoff Distribution

The natural hydrograph for the Minto Creek catchment can vary considerably from year to year. This variability is an important consideration when designing conveyance structures and storage reservoirs for shorter-term storm events. However, the specific daily or monthly runoff distribution does not significantly affect Minto's site-wide water balance during operations and closure.

Because Minto's site-wide water balance is relatively unaffected by short-term runoff events, a fixed monthly runoff distribution was adopted for the water balance model. The fixed monthly runoff distribution used in the water balance model was based on observed monthly flows rates at the hydrometric stations at W3 and W1 in Minto Creek. Figure 2-7 shows the measured runoff distributions at W3 and W1 as well as the distribution that was used in the model.



Source:Z:\01_SITES\Minto\!020_Site_Wide_Data\Water_and_Load_Balance_Files\02_Hydrology_and_Meteorology\Hydrology_Data\Aggregated_Runoff_Data_1CM002.003_Rev01_TC

Figure 2-7 Runoff Distribution Model

2.2.6 Groundwater

Assumptions for groundwater flows from the open pits were based on a revised version of the 2015 Groundwater Model (SRK 2018). Construction details for the model are presented in SRK (2015a). Pit lake water levels were updated for the revised model to reflect current pit level elevations and final pit elevations based on current mine plans:

- Current Pit Levels: Main Pit = 786.28 masl; Area 2 Pit = 765 masl
- Final Pit Levels: Main Pit = 786.28 masl; Area 2 Pit = 799 masl

No other changes were made to the model. The model water balance was queried to determine flows into and out of the pits, and groundwater discharge to surface water features. Table 2-10 summarizes groundwater flows from the open pits to several stations from the revised model.

Table 2-10 Groundwater Flows from Source to Station

Station	Total Groundwater Flow to Station		Total Inflow from Pit Lakes to Station	
	Current Pit Levels	Final Pit Levels	Current Pit Levels	Final Pit Levels
	(m3/d)	(m3/d)	(m^3/d)	(m^3/d)
WSPu	176.36	208.18	29.55	63.66
WSPdg	269.45	304.26	29.98	65.07
W3	338.37	373.09	29.99	65.12
W46	758.19	786.52	29.99	65.12
MC1	1384.70	1404.09	29.99	65.12
YR	1836.00	1844.42	29.99	65.12
Component of Total Inflow to Station from Source				
WSPu			16.8%	30.6%
WSPdg			0.2%	0.5%
W3			0.0%	0.0%
W46			0.0%	0.0%
MC1			0.0%	0.0%
YR			0.0%	0.0%
Percentage of Total Flow from Source Reaching Station				
WSPu			22.7%	49.0%
WSPdg			0.3%	1.1%
W3			0.0%	0.0%
W46			0.0%	0.0%
MC1			0.0%	0.0%
YR			0.0%	0.0%

2.2.7 Reservoirs

Reservoirs on the Minto Mine site include the Main Pit, the Area 2 Pit, and the future Ridgetop North Pit Tailings Management Facilities (TMFs) and the Water Storage Pond. Historically, the dry-stack tailings were also considered a reservoir because of the pore water stored in the tailings mass. In the model, the available storage capacities of the open pit TMFs are defined as the volume that can be filled with water, waste rock and tailings solids. Requirements for freeboard and contingency storage are not considered to be available storage and are therefore not included in values for available storage capacity.

Main Pit

The Main Pit was mined between 2007 and 2011. Since completion of mining, the pit has been used for storage of mine water and for deposition of waste rock and tailings. The estimated storage capacity of the Main Pit above the August 2016 tailings bathymetry surface and below the spill elevation of 786 m is 917,000 m^3 (Minto, 2017). In October 2017 approximately 576,000 m^3 of free mine water was stored in the Main Pit along with approximately 2,500,000 m^3 of bulk saturated tailings and 1,136,000 m^3 of bulk saturated waste rock (Minto, 2017). During the operations phase, reclaim water for processing of ore is sourced from the Main Pit TMF. After the end of the milling operation, excess water will be diverted to the Area 2 Pit until the pit is full. At that time, the Main Pit TMF outflow will be directed towards the lower Minto Creek via constructed channels that will be built as part of the closure water conveyance system.

Area 2 Pit

Development of the Area 2 Pit commenced in 2011 and mining is expected to be completed in 2018. The available storage capacity of the Area 2 Pit was approximately 3,700,000 m^3 as of early 2015 when mining of Stage 2 was complete and will be approximately 7,800,000 m^3 following the completion of Stage 3 in 2018. At the end of the mine life in 2018 it is anticipated that approximately 5,200,000 m^3 of bulk tailings and 50,000 m^3 of waste rock will be stored sub-aqueously in the Area 2 Pit.

Following the operations phase, the Area 2 Pit will be allowed to fill with water. Water from the Main Pit TMF will be directed to the Area 2 Pit through the filling phase. When full, water discharging from Area 2 Pit will flow through the primary water conveyance system via constructed channels toward lower Minto Creek.

Area 118 Pit

The Area 118 Pit is relatively small with estimated storage capacities of approximately 210,000 m^3. The pit is expected to be filled in with overburden during mining of Area 2 Stage 3 Pit. Therefore, Area 118 Pit is not considered to be a reservoir for storage of mine water or runoff.

Ridgetop North Pit

Development of the Ridgetop North Pit is planned for early 2018 and is expected to be completed in 2019. The storage capacity of the Ridgetop North Pit will ultimately be approximately 1,900,000 m^3. Tailings deposition in the Ridgetop North Pit will commence as soon as development is completed and will stop when the pit is full of tailings. The Ridgetop North Pit TSF will be covered in the dry at closure as a local high elevation to prevent a pit lake from forming.

Ridgetop South Pit

The Ridgetop South Pit is relatively small and expected to be filled in with overburden during mining. Therefore, Ridgetop South Pit is not considered to be a reservoir for storage of mine water or runoff.

Minto North

Development at the Minto North Pit began in 2015 and open pit mining is now complete. During that time, mine water was collected in the pit and pumped to Main Pit TMF. Underground development in the Minto North Pit is planned for mid-2020 to late-2021. When underground mining is complete the pit will be allowed to fill; a pit lake is expected to form (based on baseline groundwater levels in the area) but the influence of groundwater losses via pit wall fractures will only be known through performance monitoring once the pit is completed. There are no plans to store tailings or waste rock in the Minto North Pit, or to actively use the pit as a reservoir for water management purposes.

Water Storage Pond

The WSP was constructed in 2007 and has a maximum water storage volume of 320,000 m^3. Initially, the WSP was used as a source of process water for the mill. However, when the Main Pit was converted to a tailings management facility following completion of mining, reclaim from the pit supplied water for the mill, the WSP transitioned to function as a storage reservoir for clean runoff and treated water destined for release to Minto Creek. During Active Closure, the Water Storage Dam will be lowered and the footprint of the existing WSP will be converted to a Constructed Wetland Treatment System and a parallel High Flow Bypass Channel (Contango 2016).

2.2.8 Underground

Minto South Underground

Development of the Minto South Underground began in 2012 and mining is expected to be completed in 2018. Any water collected in the underground is pumped or trucked to the surface to either the Main Pit or the Area 2 Pit. Dewatering of the Minto South Underground will continue until development of the other underground areas, excluding the Minto North Underground, is complete.

Minto East Underground

Development of the Minto East Underground is planned for early 2018 and mining is expected to be completed in early 2019. Any water collected in the underground is pumped or trucked to the surface to either the Main Pit or the Area 2 Pit. Dewatering of the Minto East Underground will continue until development of the other underground areas, excluding the Minto North Underground, is complete.

Copper Keel Underground

Development of the Copper Keel Underground is planned for early 2018 and mining is expected to be completed in late 2021. Any water collected in the underground is pumped or trucked to the surface to either the Main Pit or the Area 2 Pit. Dewatering of the Copper Keel Underground will continue until development of the other underground areas, excluding the Minto North Underground, is complete.

Minto East 2 Underground

Development of the Minto East 2 Underground is planned for early 2019 and mining is expected to be completed in late 2021. Any water collected in the underground is pumped or trucked to the surface to either the Main Pit or the Area 2 Pit. Dewatering of the Minto East 2 Underground will

continue until development of the other underground areas, excluding the Minto North Underground, is complete.

Minto North Underground

Development of the Minto North Underground is planned for mid-2020 and mining is expected to be completed in late 2021. Any water collected in the underground is pumped or trucked to the surface to either the Main Pit or the Area 2 Pit.

2.2.9 Tailings Management Facilities

Minto halted deposition of tailings to the Dry Stack Tailings Storage Facility in November 2012 and transitioned to sub-aqueous deposition of tailings in the Main Pit TMF. From now until the end of mine life, tailings will primarily be deposited such that they will be saturated over the long term, with a partial volume forming unsaturated beaches in the Main Pit TMF and Ridgetop North TMF. The water and load balance model incorporates projected water volumes associated with tailings slurry, with mill reclaim water, and with pore-water in the settled tailings mass based on the life-of-mine production schedule provided by Minto.

2.2.10 Waste Rock and Tailings Cover

During the active closure period a cover system will be constructed on all waste rock piles and the Dry Stack Tailings Storage Facility. The water and load balance model assumes that the cover will reduce the infiltration of mean annual precipitation from 30% to 20%. This is consistent with previous modelling efforts for the Minto project and is supported by recent soil cover modelling results (SRK, 2015b).

3 Load Balance Inputs

All geochemical sources in the load balance model, including sources representing mine components and background catchment runoff were incorporated as dissolved loadings. Historically, the model accounted for increases in parameter concentrations arising from suspended solids in mine discharge by adding a separate source term for suspended solids at the point of discharge. Although the model retains this capability, this revision only reports on dissolved constituent loadings and concentrations; this is consistent with the dissolved nature of both effluent standards and receiving environment water quality objectives in the current site water use licence (WUL QZ14-031).

Table 3-1 Summary of Load Balance Source Terms

Source Term	Units	Applies to
Background Upper and Lower Minto Creek	mg/L	Undisturbed catchments within and downstream of the Minto Mine site
Dry Stack Tailings Seepage	mg/L	Seepage from the Dry Stack Tailings Storage Facility
Main Pit TMF Unsaturated Tailings Load	mg/year	Main Pit TMF
Mill Area Loadings	mg/year	Mill Area
Minto North Background	mg/L	Undisturbed sub-catchments in McGinty Creek
Minto North Pit Loadings	mg/year	Minto North Pit

Source Term	Units	Applies to
Nitrogen Contribution	mg/L	Added to all water released from the mine to account for loadings of nitrogen species
Ore Stockpile Concentrations	mg/L	Ore Stockpile Area, Operations
Ore Stockpile Loadings	mg/year	Ore Stockpile Area, Closure
Pit Wall Loadings	mg/year	All pit walls
Tailings Slurry	mg/L	Tailings slurry supernatant
Waste Rock Loadings	mg/m^3/year	Large Waste Rock Dumps and Mill Valley Fill Expansion (Stage 1 and Stage 2)

Source: Z:\01_SITES\Minto\1CM002.024_Water_Balance_Support\!080_Deliverables\2016_RCP_Water_and_Quality_Report\020_Tables\Source_terms_Summary_REV00_SRJ_KNK

For this update of the water and load balance, the sulphate source term for the tailings slurry was reviewed. In 2017, sulphate concentrations is the Area 2 and Main Pit water increased at a greater rate than in previous years. After investigating possible causes, SRK concluded that variability in sulphate loadings from tailings slurry was the most likely cause of the concentration increases. Details of the source term review are described in Appendix D.

In order to properly represent the variability in the mill effluent water quality, the tailings slurry source term was divided into two parts: sulphide ore tailings source term and oxide ore (referred to as POX) tailings source term. Typically, just sulphide ore or a blend of sulphide ore and POX ore are processed in the mill. Elevated concentrations of sulphate in the mill effluent coincided with processing of POX. Accordingly, the tailings slurry source terms for sulphide ore and POX ore were developed based on measured sulphate concentrations in the tailings thickener when sulphide ore and POX ore were milled. The loading rates associated with POX processing are highly variable. To maintain conservatism in the model predictions, the Expected Case and Worst Case POX ore slurry source term that represents the highest sulphate loading rates was used for all model scenarios. Details of this review and source term updates are provided in Appendix D.

3.1 Minto Creek

The load balance model was developed to evaluate the potential effects of water quality parameter loadings from mine components on water quality in lower Minto Creek and McGinty Creek. Table 3-1 shows a summary of updated geochemical source terms used in the load balance. Figure 3-1 shows the allocation of source term by sub-catchments within Upper Minto Creek. Each source term represents an estimate of runoff water quality (mg/L) or parameter loadings (mg/year) contributed by a sub-catchment or mine component.

Source term estimates were generated for two scenarios described as “Expected Case” and “Reasonable Worst Case”. The Expected Case scenario is intended to represent typical geochemical loadings (including the scale of variability observed to date) while the Reasonable Worst Case represents an upper limit to water quality parameter concentrations that may be observed in the mine water on site and consequently in Lower Minto Creek.

Source terms defined as concentrations (mg/L) were incorporated in the load balance model by assigning the water quality to all flow from the corresponding sub-catchment. Alternately, source

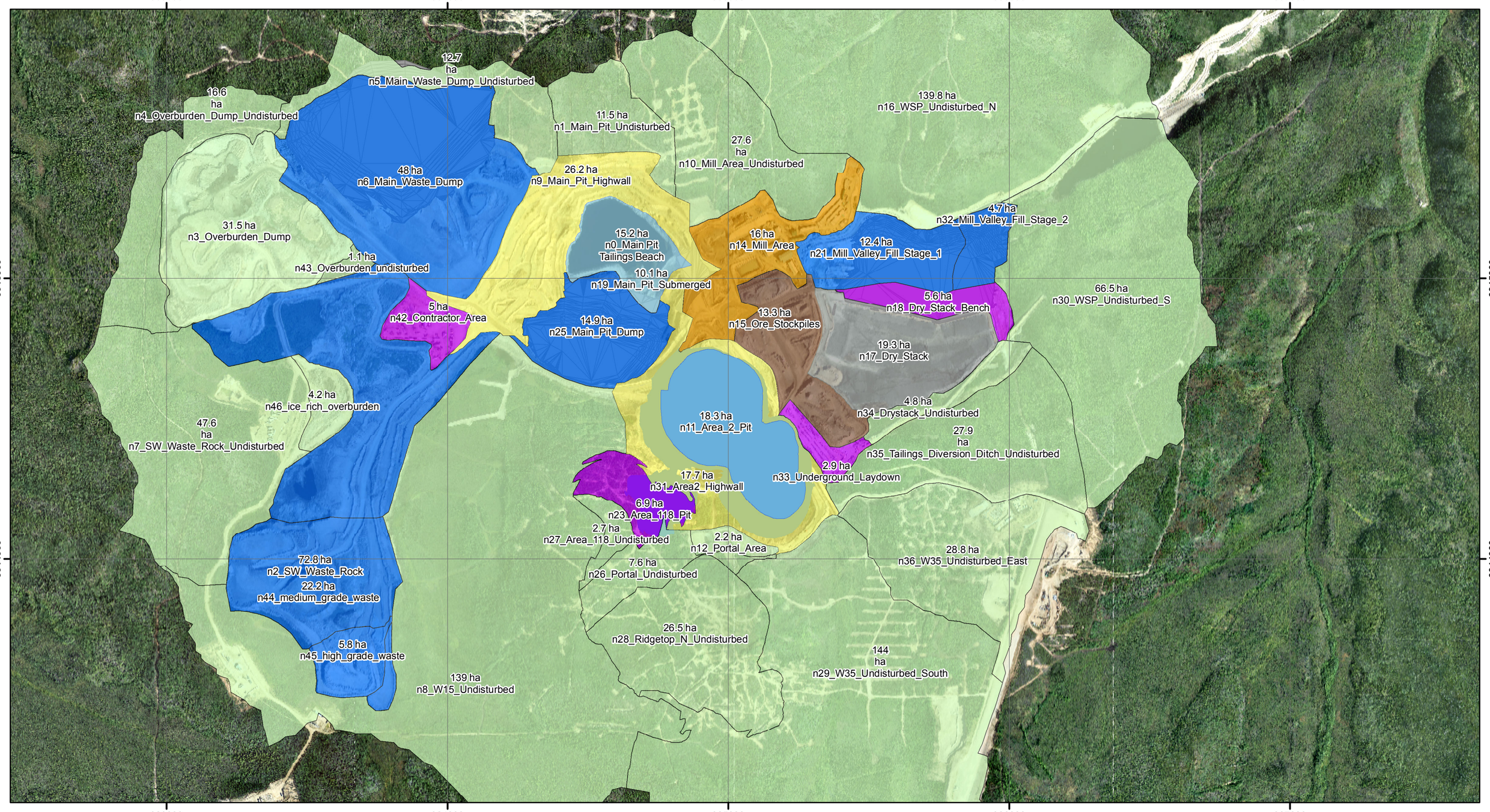
terms which were defined as loadings were added as a “dry” load to the appropriate flow or water reservoir.

Details concerning the development of source terms are described in the Minto Mine Expansion-Phase V/VI ML/ARD Assessment and Post-closure Water Quality Predictions report (SRK 2013) and in the 2016 Water Balance and Water Quality Model Summary for the Minto Mine Site provided in Appendix C. Source terms used in the model are provided in Appendix B.

All applicable source terms were updated to include water quality data up to and including December 2016 sampling events. Alexco Environmental Group conducted a review of the dataset used to update the source terms and removed erroneous data and select data points below detection (AEG 2018).

Estimates of background water quality for Minto Creek were initially developed by Minnow Environmental in 2009. Minto Creek water quality monitoring data was compiled into a pooled data set that reflected background conditions (i.e. conditions not affected by mine development activities). The data set was updated and refined in 2016 as part of a process for development of post-closure water quality objectives (WQOs). Details can be found in the Background Water Quality report (Minnow 2016). Post-closure WQOs, based on the summary statistics calculated by Minnow, have been defined by Minto. Details regarding the Post-closure WQOs can be found in the Minto Mine 2018 Reclamation and Closure Plan (RCP 2018-01(Minto 2018)).

It is important to note that both the Post-closure WQOs and background water quality source terms used in the model were derived from the same data set. However the post-closure WQOs were calculated on an annual basis and the background water quality source terms were calculated on a monthly basis. Some months, particularly in the winter, have little water quality sample data. Individual data points in these months therefore have a higher statistical weighting than the more numerous individual data points from summer months in the background water quality summary statistics. Therefore, the monthly summary statistics for background concentrations are in some cases higher than the WQOs, which are based on annual summary statistics. This is the case for dissolved copper and iron concentrations. Figure 3-2 and Figure 3-3 compare measured background concentrations to the WQOs for dissolved copper and iron. Months where background water quality source terms are derived from few data points with elevated measured background concentrations will show consistent exceedance of WQOs in the model results.



N

0 75 150 300 450 600 Meters

Legend

Minto Phase V/VI Source Term Allocation

- Background Upper Minto Creek
- Drystack Seepage
- Main Pit TMF Unsaturated Tailings Load
- Mill Area Loadings
- None
- Ore Stockpile Loadings or Concentrations
- Pit Wall Loadings
- Ridgetop TMF Unsaturated Tailings Load
- Waste Rock Concentrations
- Waste Rock Loadings

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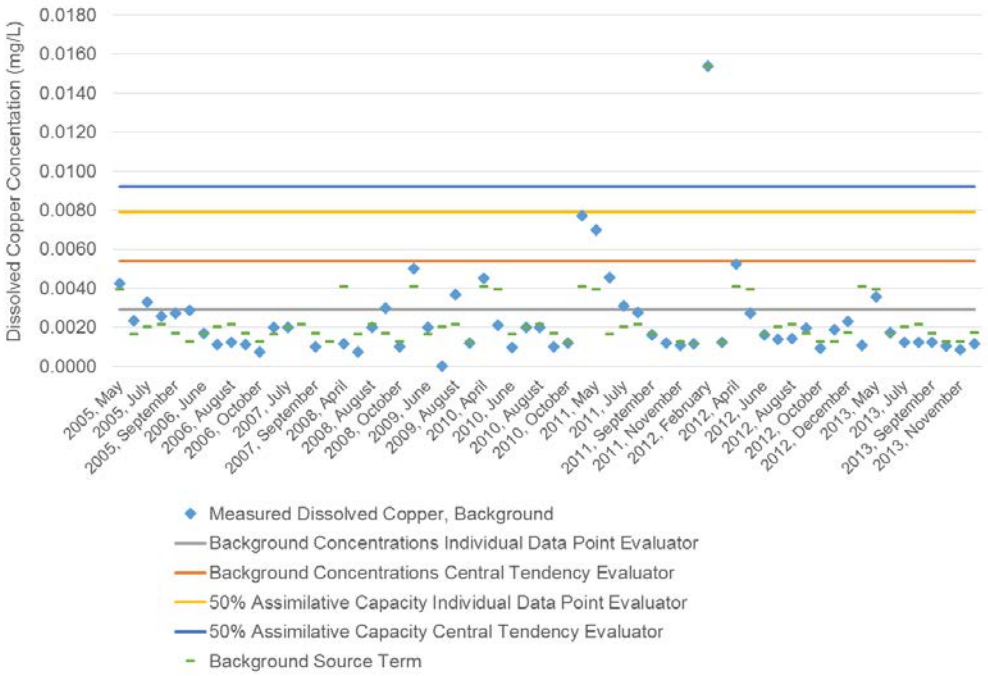
Minto Mine

Water and Load Balance Report 2016

Source Term Allocation

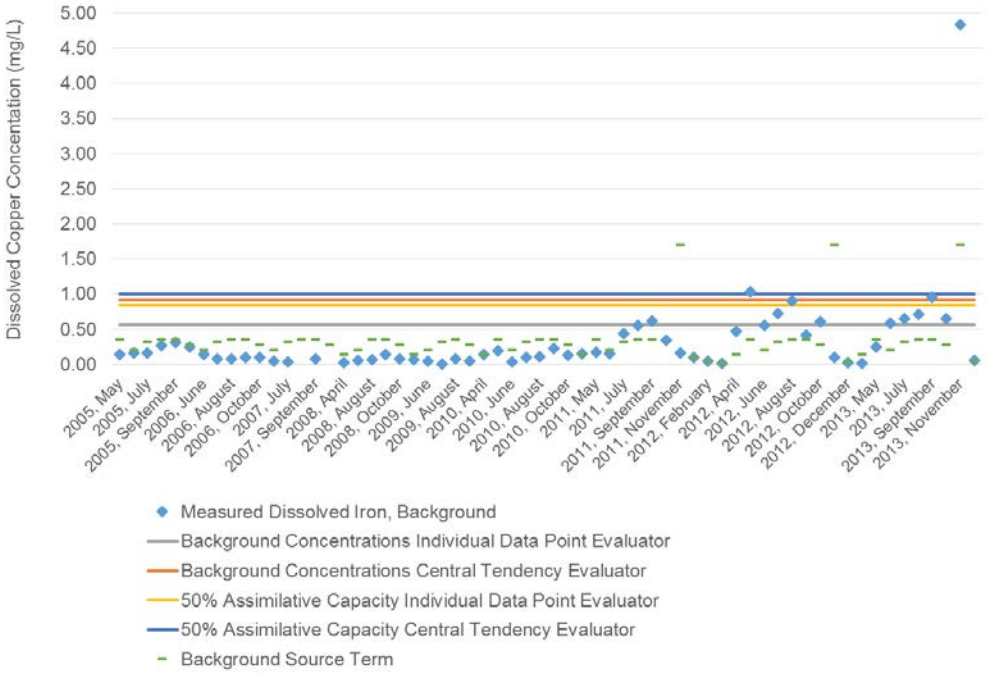
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Figure 3-2 Measured Background Water Quality and WQOs for Dissolved Copper



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Figure 3-3 Measured Background Water Quality and WQOs for Dissolved Iron

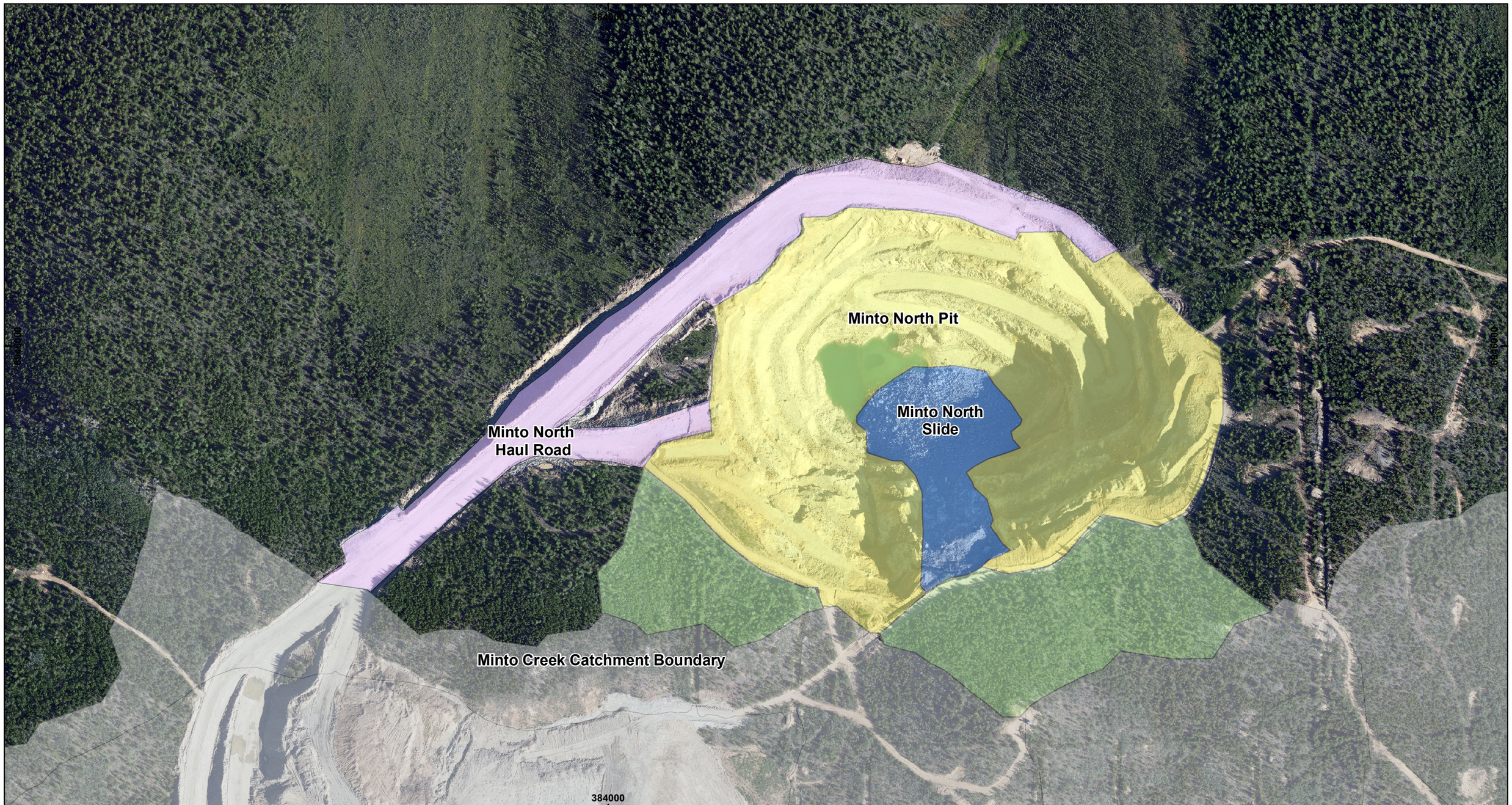
3.2 McGinty Creek

Background water quality data for McGinty Creek is available for the months of April to November. The McGinty background water quality used in the model for all undisturbed sub-catchments was the average of the water quality data at the monitoring station located in McGinty Creek upstream of the confluence with the Yukon River (Station MN-4.5).

The source terms for the Minto North Pit walls were determined based on humidity cell test results (SRK 2013). The source terms for the access road and pit wall failure were assigned the waste rock source term which were determined based on measured water quality at W15.

Figure 3-4 shows the allocation of source terms for the Minto North Pit area. In the water and load balance model, the Minto North Haul Road material was routed towards the upper-eastern catchments of McGinty Creek and from there to lower McGinty Creek. Loadings from the pit walls and slide material was assumed to report to the bottom of the open pit. The rate of seepage of mine water from the open pit to the surrounding groundwater aquifer is difficult to estimate. Therefore, a very conservative approach was adopted to estimate of the theoretical effect of groundwater seepage on water quality in McGinty Creek. It was assumed that all loadings associated with the open pit mine water would report to McGinty Creek after following a lag time of 15 years. This approach is expected to overestimate the transfer of loadings from Minto North Pit to McGinty Creek by at least one order of magnitude.

In addition, the load balance model for Minto North incorporated worst case conditions in terms of source loadings. The entire Minto North Pit wall and waste rock slide material was assumed to remain exposed in the post-closure phase (i.e. maximum exposed area acting as a loading source). The combination of maximum loading rates and maximum pit area exposed is considered to be a very conservative estimate of future loading rates. Partial flooding of the Minto North Pit or application of steady state loading rates, as observed in the humidity cell tests, would result in lower loading rates. Source terms derived for the Minto North Pit are listed in Appendix B.



N

0 20 40 80 120 Meters

Legend

- Minto Creek Catchment

Source Term

- Background McGinty Creek
- Pit Wall Loadings
- Waste Rock Concentrations
- Waste Rock Loadings

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Minto Mine

Water and Load Balance Report
Report 2018

Source Term Allocation
Minto North Pit

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4 Model Implementation

4.1 Model Version

The water and load balance model for Minto was developed using the GoldSim software package (version 11.1). The model scenarios described here were implemented in model Version #68.

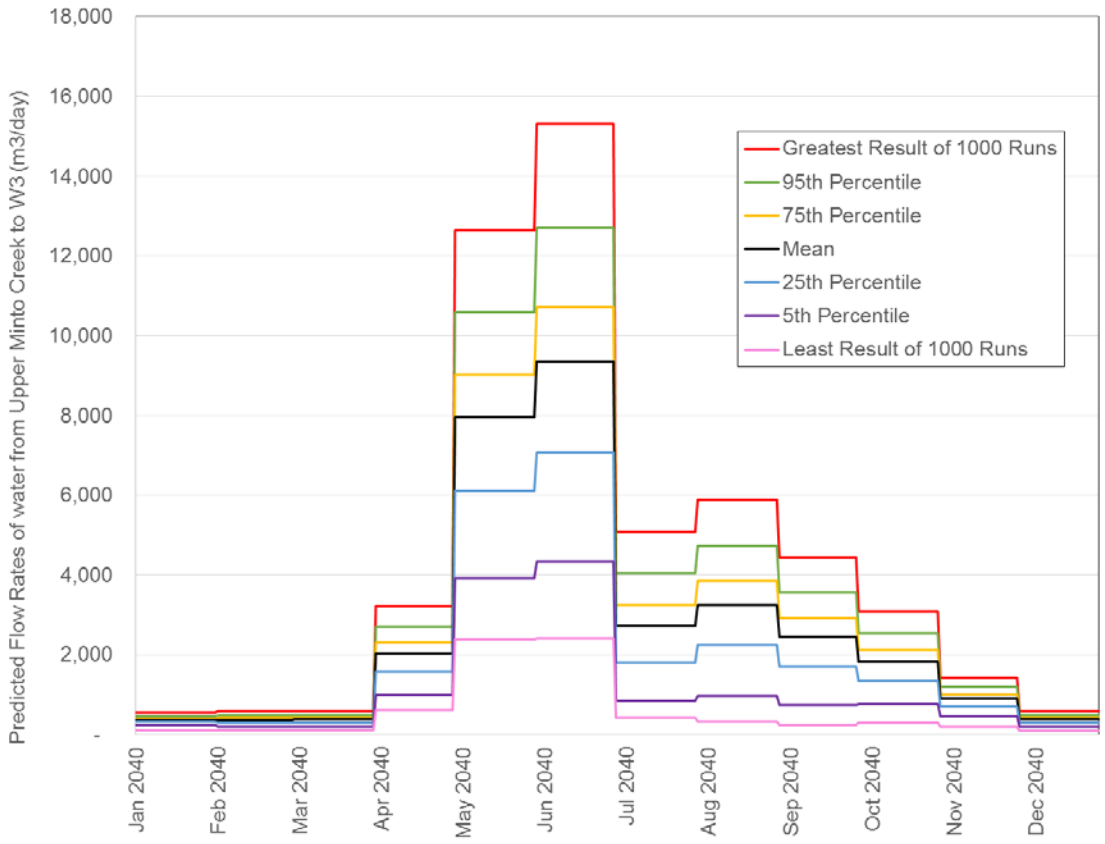
4.2 General Modelling Approach

4.2.1 Stochastic Water Balance Model

Water balance results for the 2018 water balance model were generated by running the model as a Monte Carlo simulation. Monte Carlo simulations are well suited for situations where the value of a key input cannot be predicted but where the distribution of the input is known or can be adequately estimated. Total annual precipitation for the Minto site is an example of such a variable. Although it is not possible to predict the rate of annual precipitation in any given year, it is possible to develop a probability distribution (see Section 2.2.1).

In the water balance Monte Carlo run, the model randomly selected a value for total annual precipitation from the probability distribution developed for the site (see the Pearson III distribution shown in Table 2-5, Section 2.2.1). Annual runoff volumes were then calculated by multiplying total annual precipitation by a runoff coefficient and catchment area (see Sections 2.2.3 and 2.2.4). The calculated runoff would then be distributed over all months of the year according to the typical hydrograph used in the model (described in Section 2.2.5). The model was run in this manner from year 2015 through 2050, each year with a randomly selected precipitation value, and all results were recorded and stored by the model. A total of 1,000 model runs were completed in this way.

At the end of 1,000 model runs, all results were compiled and probability distributions of the results were generated. Figure 4-1 shows an example of model results for a single year from the Monte Carlo simulation. For illustration purposes, a year well into the post-closure two period is shown to avoid having the example be affected by active water management at site or by transient conditions that correspond to filling of pits in the closure periods. The results show the possible range of flow rates from Upper Minto Creek to W3 as represented by the model. The most likely flow distribution is the mean flow (black line).



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Figure 4-1 Example of Result Generated by the Monte Carlo Simulation – Predicted Range of Average Flow from Upper Minto Creek to W3 in 2040

Routing of runoff in the model followed the plan detailed in the water management plan and closure plan as described in Section 4.3. The modelling period began on January 1, 2015 and ended on January 1, 2050. The operations period was modelled as ending at the end of April 2022, followed by the closure periods. The water and load balance model assumed that all closure activities were implemented instantaneously at the end of operations.

4.2.2 Loading Balance Model

Loadings were incorporated in the model by associating loadings source terms with the corresponding water flows or mine components as follows:

- Concentration based source terms were applied as constant values to monthly runoff volumes from corresponding sub-catchments as described in Section 3.
- Loading based source terms were incorporated into the model as a “dry” load either to runoff or to water reservoirs. For example, loadings from tailings solids were applied to the water in the reservoir where tailings were deposited.

The development of loadings source terms is discussed in Section 3. The source terms have been updated periodically as new data becomes available. The most recent source term update is described in Appendix B.

4.3 Water Management

The focus of the water management strategy during the operations phase is to maintain a minimal but adequate inventory of mine water on site. The water inventory will, to the extent possible, be managed by diverting clean (i.e. discharge compliant) surface runoff away from developed mine areas and towards the WSP to potentially be released to Minto Creek. The modelling representation of water conveyance and water diversion options are described in the following section.

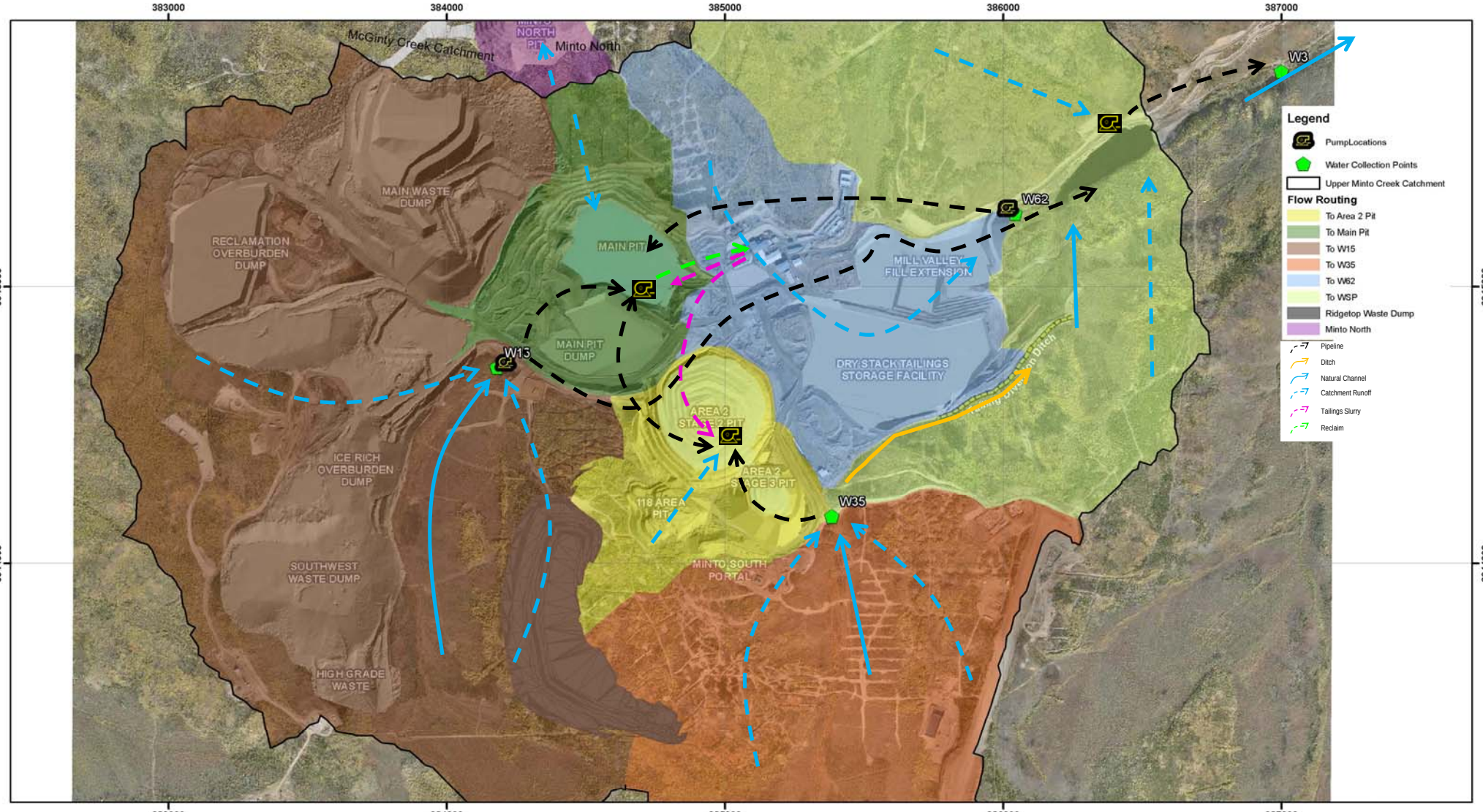
Figure 4-2 and Figure 4-3 show schematics of how water was routed in the water balance model during operations and in the post-closure phases.

During the operations period water collected at W15 is directed to the Main Pit or to the WSP through the piping network on site. Runoff collected by the W35 channel is directed to the Main Pit or to the WSP via a pipeline. Mine water collected at W62 is pumped to the Main Pit. Slurry tailings were deposited in the Main Pit until April 2015 and in the Stage 2 area of the Area 2 Pit thereafter. Tailings are planned to be deposited in the Area 2 Pit Stage 2 and Stage 3 until development of the Ridgetop North Pit is complete. At that time tailings will be deposited in the Ridgetop North Pit until it is full at which point tailings will be deposited in the Area 2 Pit. All mill reclaim water is sourced from the Main Pit throughout the operations period.

The model assumes that the mine enters the post-closure phases immediately following the end of mine life in April 2022. During post-closure (Figure 4-3), water conveyance structures were assumed to be constructed to direct runoff as follows:

- Water collected at W15 reports to the Main Pit along with runoff from upstream catchment areas.
- The Main Pit outflow reports to the Area 2 Pit, until the pit has filled with water and a post-closure pit lake has formed (not shown on Figure 4-3).
- When the Area 2 Pit Lake has formed, the outflow from the Main Pit area reports to the former mill site area, through a channel along the mine access road to the WSP Wetland.
- Catchments upstream of the Area 2 Pit reported directly to the Area 2 pit lake.
- Water in the Area 2 Pit flows via a spill-way through the former mill site and through a channel along the mine access road to the WSP Wetland.
- Runoff from the former mill area and from slopes along the Mill Valley reported directly to the WSP Wetland.
- Water treated by the WSP Wetland report directly to Minto Creek.

Table 4-1 provides further details concerning the implementation of water routing in the water balance model.



Notes: catchment delineation by Access Consulting Group (2013)

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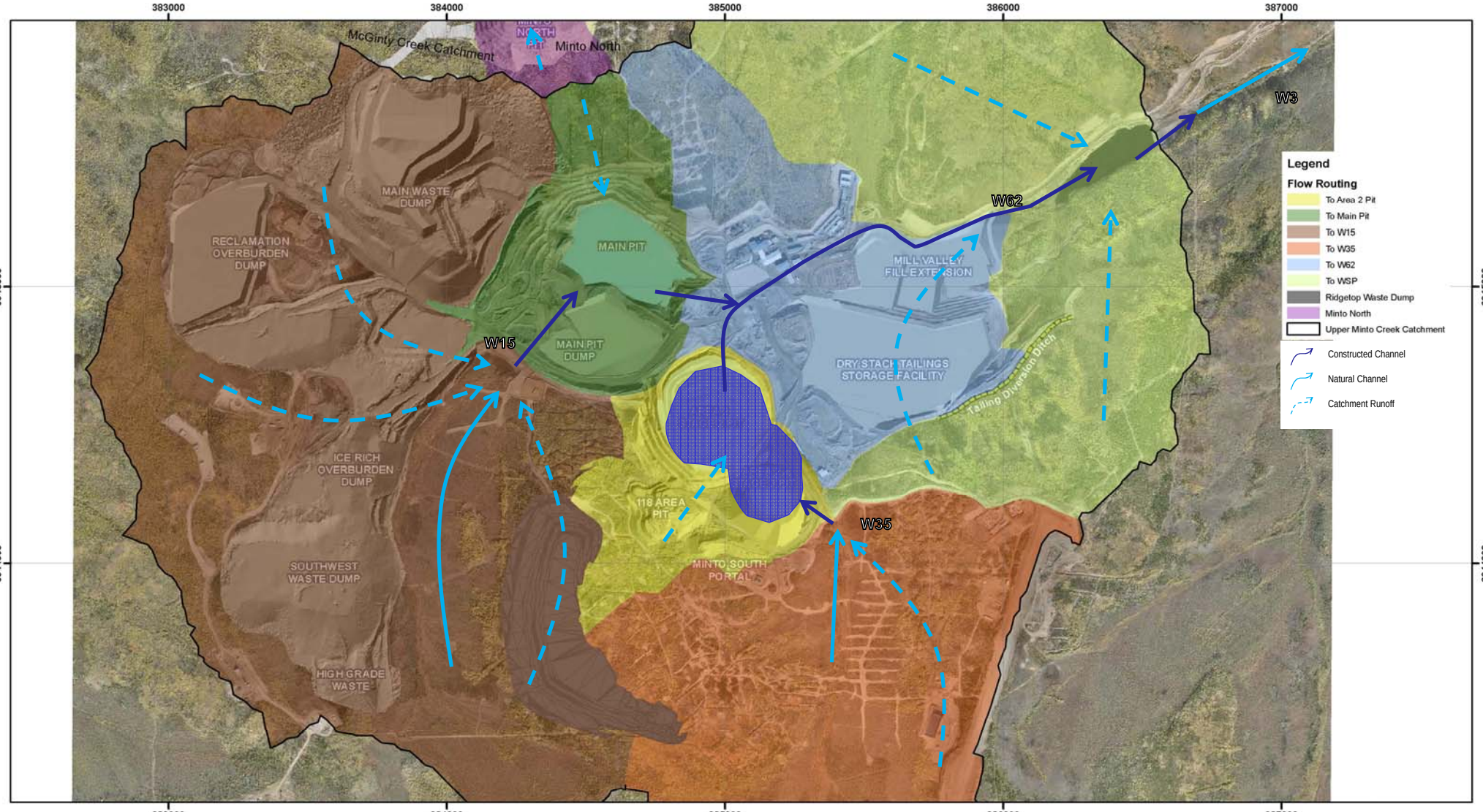
Minto Mine

Water and Load Balance Report 2018

Modelled Routing of Water: Operations

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0 100 200 400 600 800 Meters

Notes: catchment delineation by Access Consulting Group (2013)

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Minto Mine

Water and Load Balance Report 2018

Modelled Routing of Water: Post Closure

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Table 4-1 Modelled Routing of Runoff and Mine Water Conveyance

Input	Period	Reservoir	Output	Period
Catchment Yield	Always	**WSP (WSP Wetland as of 2023)**	Minto Creek	Always
W15	Until 2022			
W35	Until 2022			
Area 2 Pit	2025 onwards			
W62	Until 2026	**Main Pit**	to Area 2	2018 to 2024
Catchment Yield	Always			
Minto North Pit	2015 to 2016		Evaporation	Open water seasons
W15	Always			
			WSP Wetland	2026 onwards
Tailings Slurry Water	Intermittently 2015 to 2016 & in 2018		Reclaim Water	2015 to 2022
Catchment Yield	Always	**Area 2 Pit**	Main Pit	Intermittently 2015-2022
W35	2022 onwards		Evaporation	Open water seasons
Main Pit	2018 onwards			
Tailings Slurry Water	Intermittently 2015 to 2022		WSP Wetland	2025 onwards
			Reclaim Water	Intermittently 2020-2022
Catchment Yield	Until 2022	**Ridgetop North Pit**	Area 2 Pit	2018 to 2022
Tailings Slurry Water	2019 to 2022		W35	2022 onwards
Catchment Yield	Always	**Minto North Pit**	to Main Pit	2015 to 2016
			to McGinty Creek	2017 onwards

4.4 Tailings Slurry Deposition

Tailings slurry deposition was implemented in the water balance as follows:

- Q1 2015 to Main Pit TMF
- Q2 2015 to Main Pit TMF and Area 2 Pit TMF
- Q3 2015 to Q2 2019 to Area 2 Pit TMF
- Q2 2019 to Q1 2021 to Ridgetop North Pit TMF
- Q1 2021 to Q1 2022 to Area 2 Pit TMF

The dry bulk tailings density was assumed to be 1.35 tonnes/m^3. Details concerning tailings slurry deposition for the operations period were provided by Minto.

4.5 Model Calibration

4.5.1 Water Balance Model Calibration

The approach used to calibrate the water model was a site wide yield coefficient that correlated the total annual precipitation to the total annual yield. This approach avoids the uncertainties inherent in quantifying appropriate input values for individual parameters (evaporation, evapotranspiration, sublimation, etc.) by determining a single site wide yield coefficient based on measured values of both precipitation and catchment yield (stream flows) from historical records. The yield coefficient approach integrates all catchment processes into a single empirical factor that describes the relationship between precipitation and catchment yield.

In years for which the total annual precipitation was known, the actual (measured) total annual yield was compared to the modelled total annual yield. The modelled data was calibrated to the measured data by adjusting the yield coefficient to achieve a best fit. After a suitable yield coefficient was identified, it was used to model future site yield estimates.

4.5.2 Loading Balance Model Calibration

The load balance was calibrated by comparing constituent concentrations in the Main Pit and Area 2 Pit water to model predictions and then adjusting source terms accordingly to improve the fit between measured and modelled concentrations. The main sources of loadings to pit water is waste rock seepage and tailings slurry. Source terms for waste rock seepage are based on measured seepage water quality and are therefore, in a sense, already calibrated. Also, waste rock seepage water quality is not expected to change abruptly because of the size of the waste rock areas. Loadings from tailings slurry, on the other hand, are directly related to the characteristics of the ore processed, which can change in short order as the ore type changes. Therefore, the load balance calibration was focused on the ore source term.

The tailings slurry source term update sought to address some of the variability inherent with tailings slurry water quality by generating multiple source terms that characterize sulphide ore tailings slurry and POX ore tailings slurry. These source terms were generated using water quality data from the tailings thickener from 2015 to 2017 as well as the sulphide ore and POX ore milling schedules. Details regarding the updated tailings slurry source terms can be found in Appendix D. In general, the model predictions matched the measured concentrations in the Main Pit and Area 2 Pit reasonably well so calibrations were not needed.

4.6 Water Treatment

4.6.1 Active Treatment

The water treatment plant at Minto can be operated in different configurations depending on the quality of feed water to be treated. Treatment options range from simple TSS removal to sulphide precipitation to reverse osmosis (RO) treatment, which removes 95% to 99.5% of all constituents in the mine water.

In the load balance model scenarios, it was assumed that the water treatment plant would not be operated. This assumption is conservative and allows the effectiveness of the passive water treatment options to be assessed independently.

4.6.2 Passive Treatment

The Minto Mine plans to construct a wetland where the WSP is currently located for long-term passive treatment of water flowing from the mine site during the post-closure phases. The WSP wetland is expected to remove a variety of water quality constituents as described in (Contango 2017). In the load balance model, it was assumed that the wetland will be operational at the start of 2027, approximately 5 years from the end of mine life. The actual commissioning date of the WSP wetland will be contingent on the wetland testing phase that will begin once the wetland has been constructed after the end of mine life. Once the wetland is operational and treats water to an acceptable standard all active water treatment facilities will be decommissioned.

Implementation of the wetland in the water and load balance model was based on the preliminary design basis and operational philosophy. Accordingly, the model assumes that the WSP Wetland will have a total active volume of approximately 2,000 m^3 and a maximum inflow of 3,000 m^3/day. This results in a minimum Hydraulic Retention Time (HRT) of about 16 hours. Should the flow rate into the WSP Wetland exceed 3,000 m^3/day, excess water will be diverted around the wetland untreated. Estimates of removal rates for dissolved cadmium, copper and selenium were incorporated in the model. The estimates rates were based on results of the wetland testing program carried out by Minto (Contango 2016). The estimated removal rate equations are shown in Table 4-2.

Table 4-2 WSP Wetland Load Balance Dissolved Metal Removal

WQ Constituent	Rate Equation	Order	Removal Limit
Ammonia-N	$[NH_3]e^{-0.143/day*HRT}$	First	MDL
Cadmium – Cd	$[Cd]e^{-1.14/day*HRT}$	First	0.000009 mg/L
Copper – Cu	$[Cu]e^{-1.17/day*HRT}$	First	0.015 mg/L
Molybdenum - Mo	$[Mo] - \frac{0.0006\frac{mg}{L}}{hr} * HRT$	Zero	0.005 mg/L
Nitrate-N	$[NO_3]e^{-0.591/day*HRT}$	First	MDL
Selenium - Se	$[Se]e^{-0.319/day*HRT}$	First	0.001 mg/L
Zinc - Zn	$[Zn]e^{-1.14/day*HRT}$	First	0.030 mg/L

Notes:

(1) HRT = Hydraulic Retention Time in hours

(2) MDL = Method Detection Limit

(3) [Cd] = Concentration of dissolved cadmium in WSP Wetland Influent in mg/L

(4) [Cu] = Concentration of dissolved copper in WSP Wetland Influent in mg/L

(5) [Se] = Concentration of dissolved selenium in WSP Wetland Influent in mg/L

The modelled performance of the WSP Wetland was not explicitly adjusted for hydraulic efficiency. Rather, hydraulic efficiencies were assumed to be on the same order of magnitude (or higher) than

hydraulic efficiencies associated with the pilot-scale or field scale wetlands that were used as a basis for estimating removal rates.

5 Model Results

5.1 Water Balance Results

5.1.1 Minto Creek Water Balance Results

The primary water management challenge for the Minto Mine is to maintain sufficient storage capacity to meet the anticipated life-of-mine storage requirements and that sufficient stored water is available to meet the needs of the milling operation.

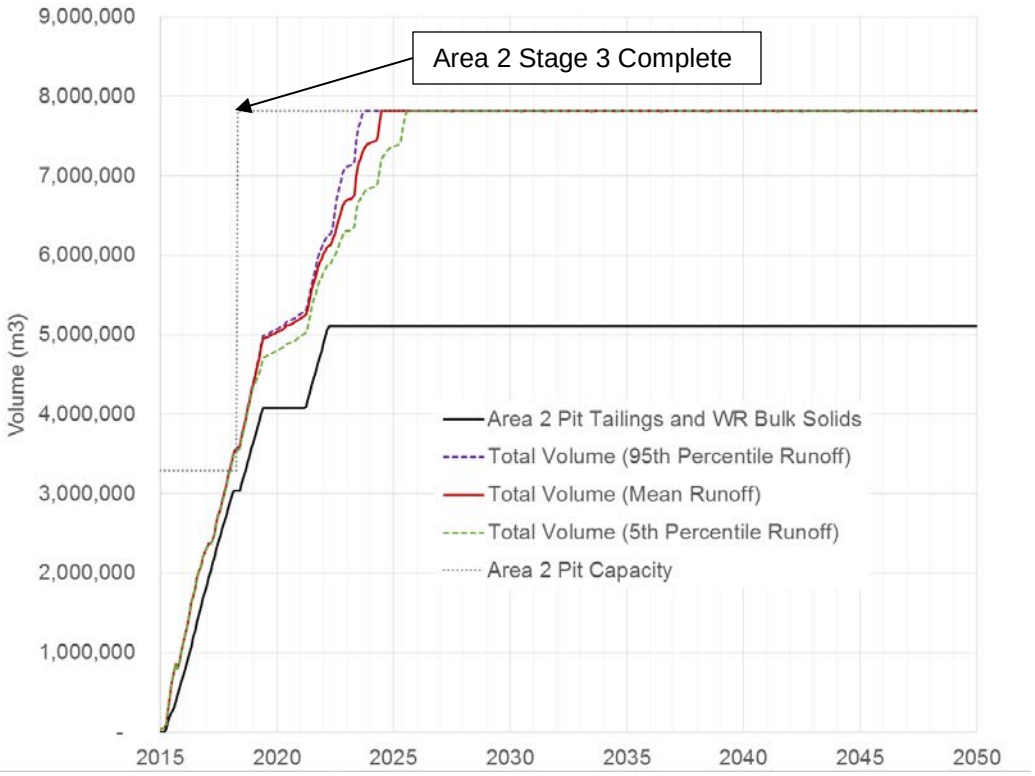
The volume of water that must be released from site each year to maintain an appropriate water inventory is an important factor for managing water on site. By quantifying the volume of water that can be added to the inventory in each year, and consequently the volume of water that should be released, water management staff can implement diversions or plan water treatment campaigns, if required.

Figure 5-1 shows projected volumes of tailings and free mine water in the Area 2 Pit through the operations and post-closure periods for the modelled scenarios. Note that, in the modelled scenarios, filling of Area 2 Stage 2 Pit reaches the saddle elevation prior to the completion of mining in Area 2 Stage 3-4. At that time tailings deposition will be redirected to the Main Pit TMF until Area 2 Stage 3-4 is complete and available for tailings storage. The results indicate that, under the mean runoff model scenario, the Area 2 Pit will be full of water by mid-2024.

5.1.2 McGinty Creek Water Balance Results

The water balance for the Minto North Pit consists of inflows from a small upgradient catchment and from direct precipitation on the open pit. During development and mining of the pit the mine water is being transferred to the Main Pit in the Upper Minto Creek catchment. After mining is complete, the open pit will be allowed to fill. Although it appears that there is a slight net positive water balance for the pit (including the contributions from the upgradient catchment), it is expected that it will take years for the pit to fill and discharge as surface flow to lower McGinty Creek, if it fills at all. Depending on the rate of groundwater flow into and out of the pit area, it is possible that the pit may never discharge via surface overflow (as noted in Section 2.2.7). At this stage of the project, it is not possible to definitively predict whether a lake will form, and what steady state water surface elevation will be established if a lake does form.

The catchment area influenced by the pit is a small component of the overall McGinty Creek catchment (roughly 15 ha vs. 3,400 ha) and therefore the Minto North development is expected to have minimal (i.e. likely not measurable) effect on flow volumes in McGinty Creek.



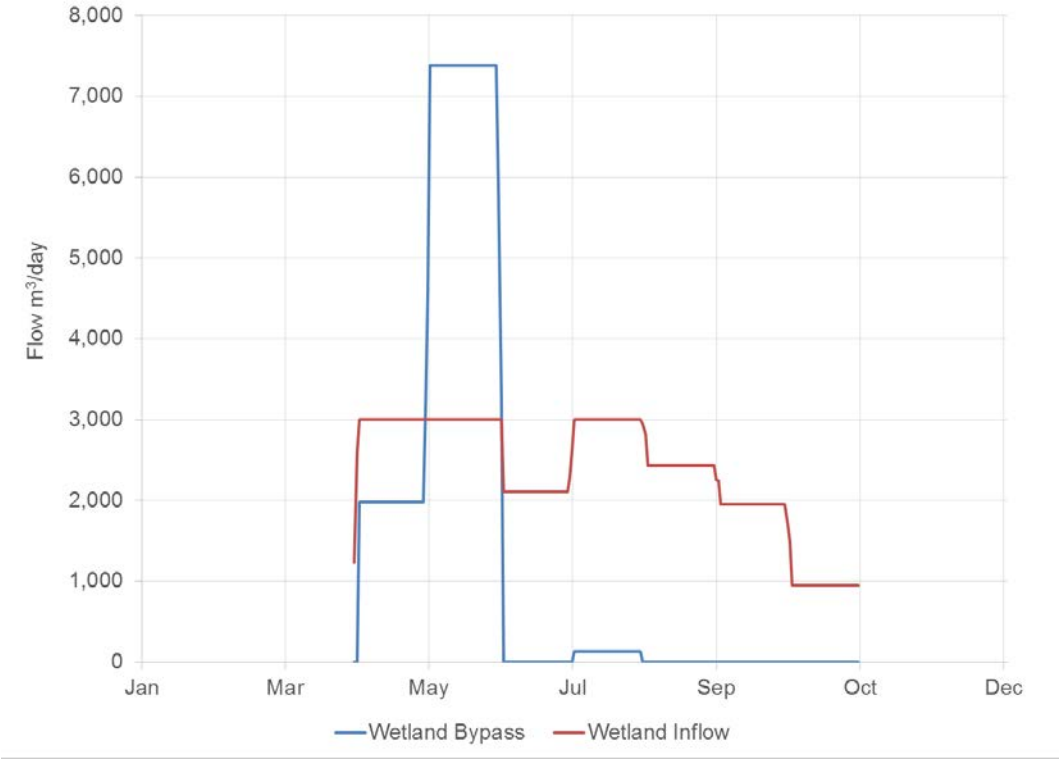
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Figure 5-1 Modelled Area 2 Pit Water and Tailings Volumes

5.1.3 Constructed Wetland Water Balance Results

The design treatment capacity of the wetland is 3,000 m^3/day. The flow rate through the proposed wetland system and high-flow bypass structure during a typical year are shown in Figure 5-2.

As discussed in Section 4.2.1, the stochastic water balance model results produced by the Monte Carlo simulation are probability distributions of water balance outcomes. The average values presented Figure 5-1 are the most likely outcomes, but a range of outcomes are possible. In the results for the post-closure periods when the Area 2 Pit is full, the modelled discharge volumes change with annual rates of precipitation but there is no longer any net change in water storage on site.



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Figure 5-2 CWTS and High-Flow Bypass Typical Monthly Flow Rates

5.2 Water Quality and Parameter Loadings Results

5.2.1 Post-Closure Water Quality Objectives

Water Quality Objectives (WQOs) have been developed by Minnow Environmental and agreed upon by Minto Explorations and the Selkirk First Nations. Details of the development and application of the WQOs is outlined in the RCP (Minto 2018). In general, the WQOs are summarized as follows:

1. *Closure WQOs that meet the criteria for non-degradation of background concentrations in Minto and McGinty creeks.*

2. *If non-degradation of water quality is not achievable, the closure WQOs shall be set at no more than 50% of the assimilative capacity of Minto Creek using the following formula:*

Closure WQO = BKGD + (0.5[WQO Operations – BKGD])*

where: BKGD = background concentration

The post-closure 50% assimilative capacity WQOs, as calculated using the above equation utilizing the background water quality summary statistics provided by Minnow (2016), are included in the water quality results for station W1 for comparison purposes. A complete list of the Post-closure WQOs is provided in Minto (2018).

5.2.2 Minto Creek Water Quality Results

Water quality predictions for RCP 2018-01 were produced for the five scenarios listed in Table 5-1.

Table 5-1 Model Scenarios

Scenario	Source Term	Phase VII Developments	Passive Treatment
1	Expected Case	Included	None
2	*Reasonable Worst Case*	Included	None
3	*Reasonable Worst Case*	Not Included	None
4	Expected Case	Included	Included
5	*Reasonable Worst Case*	Included	Included

All scenarios include the effect of soil covers as described in Section 2.2.10.

The *Reasonable Worst Case* is a conservative case while the *Expected Case* can be considered to be the most likely outcome, based on the available information. The water quality results shown below include concentrations of copper, selenium, cadmium, and sulphate which currently are parameters of potential concern and are demonstrative of the effectiveness of passive treatment with the exception of sulphate. Copper and selenium are generally indicative of the behaviour of other parameters.

The five scenarios were selected to show the potential effects of source terms, the WSP wetland and the addition of the Phase VII mine developments on the quality of water immediately downstream of site (reported as concentrations at W3) as well as concentrations at W1 in Lower Minto Creek as follows:

- Scenarios 1 and 2 allow for a comparison between effects of *Expected Case* and *Reasonable Worst Case* source terms.
- Scenarios 2 and 3 compare effects on water quality with and without Phase VII mine development.
- Scenario 4 and 5 allow for comparison of wetland performance in cases where the inflow water quality has been influenced by the *Reasonable Worst Case* source terms and by *Expected Case* source terms using the load removal rate equations shown in Table 4-2.

The two model scenarios that do not incorporate passive treatment give a sense of the range of concentrations that may occur based solely on variation in geochemical loading rates from mine-related and background sources and on runoff volumes that vary seasonally with precipitation and melt. The two model scenarios that incorporate passive treatment illustrate the potential improvement in water quality that could be achieved by a constructed wetland treatment system.

Model results for water quality in pits, water released to station W3 in upper Minto Creek and at station W1 in Lower Minto Creek for the entire modelled period (2015 to 2050) are included in Appendix A1. Tabular monthly results are provided for a single year well into post-closure (2045) in Appendix A2 to illustrate typical modelled seasonal variability during the post-closure period. Results are included for open water season only (April to October) as the background water quality source terms for the winter months are highly variable and poorly characterized, and therefore produced misleading results when used as inputs for modelling.

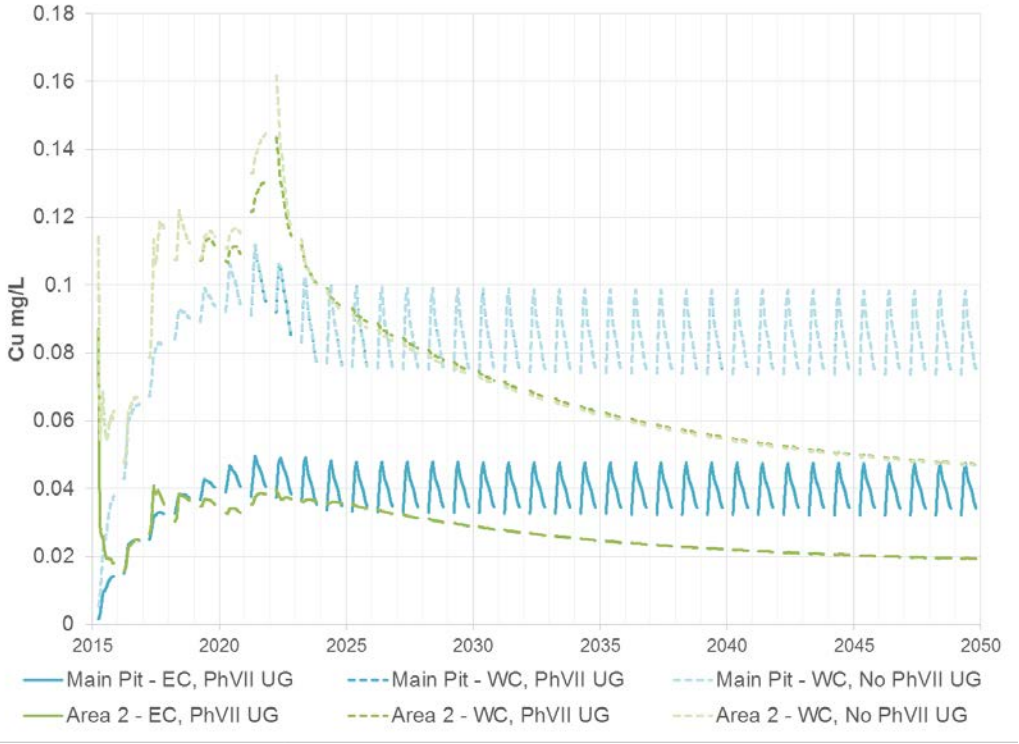
The following acronyms are used in the legends for the figures below:

- EC: *Expected Case* source terms
- WC: *Reasonable Worst Case* source terms
- PhVII UG: Includes Phase VII underground developments
- no PhVII UG: does NOT Includes Phase VII underground developments
- WT: includes wetland treatment
- no WT: does not include wetland treatment

Figure 5-3 to Figure 5-6 show predicted dissolved copper, selenium, cadmium, and sulphate concentrations for mine water in the Main Pit and Area 2 Pit. Concentrations in the pits are expected to increase during the operational stage when waste rock and tailings are deposited. Selenium concentrations increase markedly during periods when tailings are deposited to the pit. After closure, copper, selenium, cadmium, and sulphate concentrations are expected to decline to steady-state levels that are representative of long-term post-mining geochemical loading rates. Concentrations in the Main Pit remain relatively constant, again reflecting geochemical loading rates.

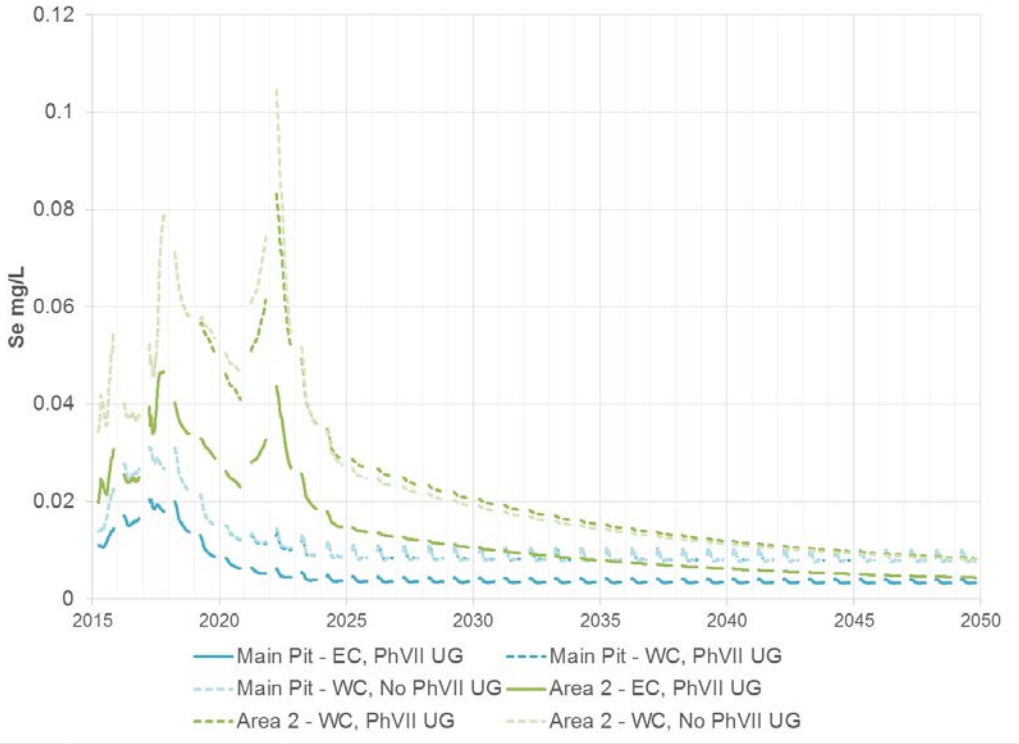
Figure 5-7 to Figure 5-10 show model predictions for dissolved copper, selenium, cadmium, and sulphate concentrations for mine water in the WSP as released to station W3 in Minto Creek. Both profiles show relatively low concentrations during the operational period followed by a modest increase in concentrations at closure. This is because residual mine water stored in the Area 2 TMF begins reporting to Minto Creek at that time.

Figure 5-11 to Figure 5-14 shows modelled values total copper, selenium, cadmium, and sulphate concentrations in Lower Minto Creek (the W1 water quality station). Copper and selenium concentrations follow the trends noted for the W3 water quality predictions but the difference between concentrations during the operations stage and post-closure periods is less pronounced because of dilution from the Lower Minto Creek catchment. Concentrations of nitrate, nitrite and ammonia (not shown) are expected to decrease gradually following the end of active mining in 2018. Residual ammonium nitrate/fuel oil (ANFO) explosives in tailings and water rock contribute the vast majority of nitrogen species to the mine water. Once mining is complete and ANFO use ends, there will be no new sources of nitrate and ammonia loadings.



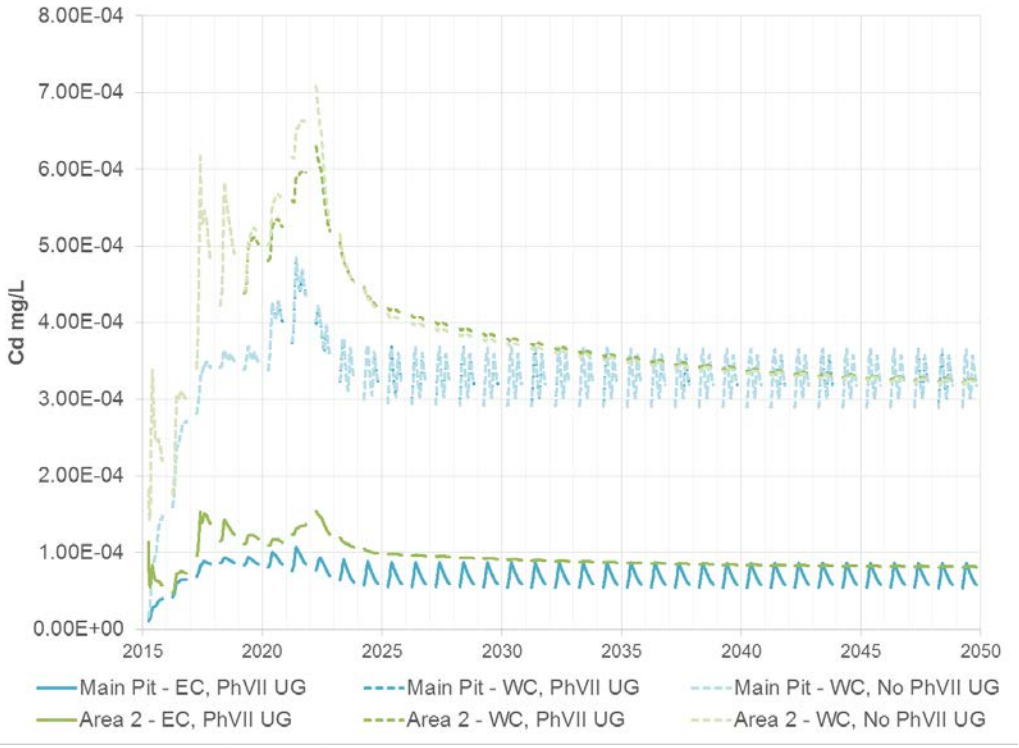
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Figure 5-3 Model Predictions of Dissolved Copper Concentrations in Pit Lake Water



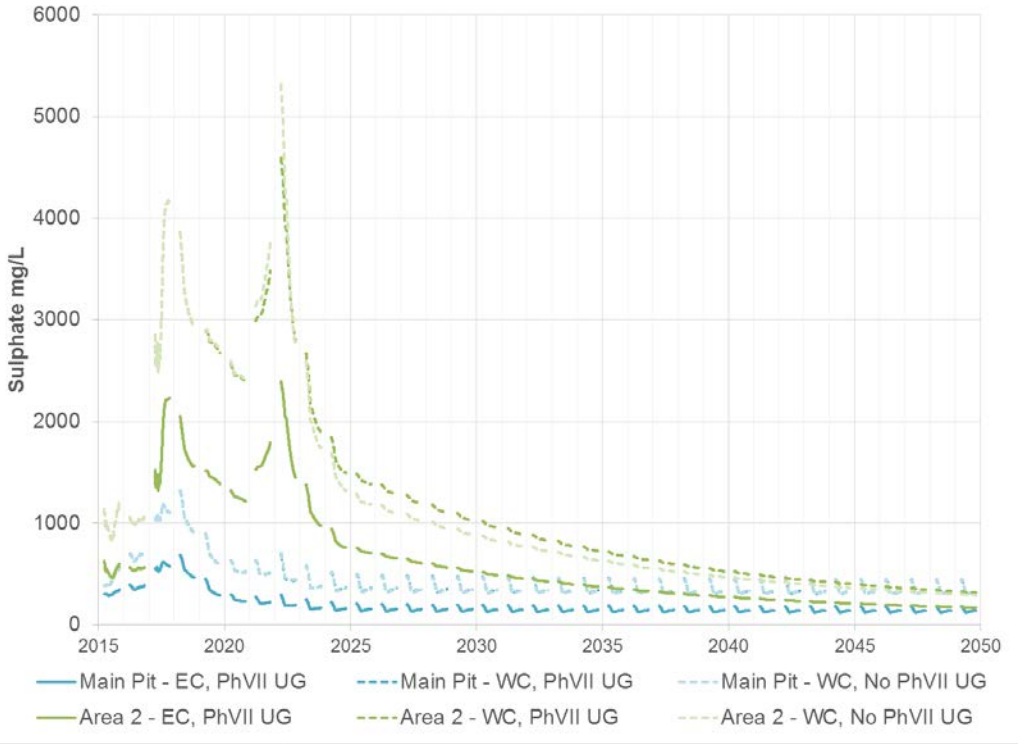
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Figure 5-4 Model Predictions of Dissolved Selenium Concentrations in Pit Lake Water



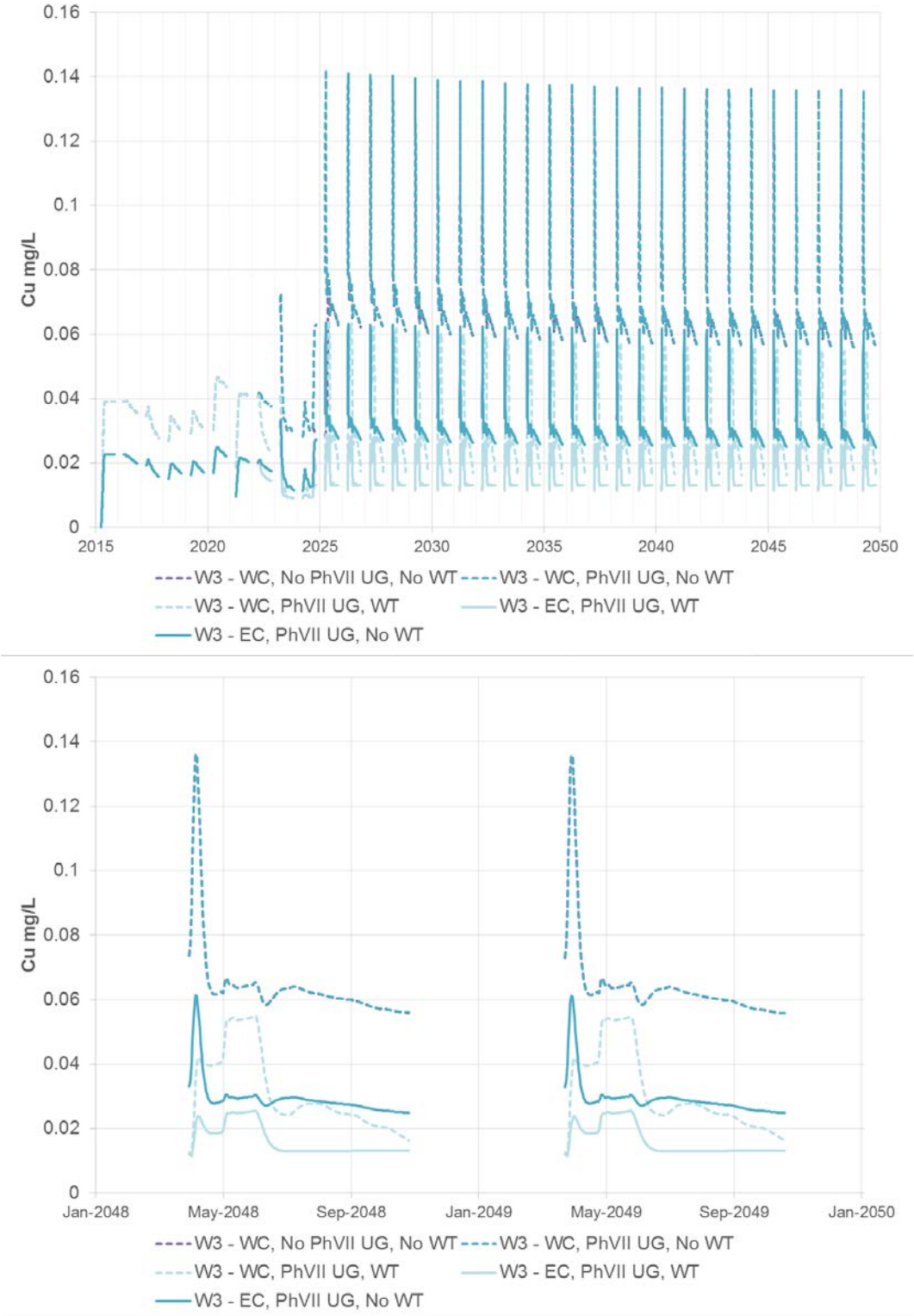
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Figure 5-5 Model Predictions of Dissolved Cadmium Concentrations in Pit Lake Water



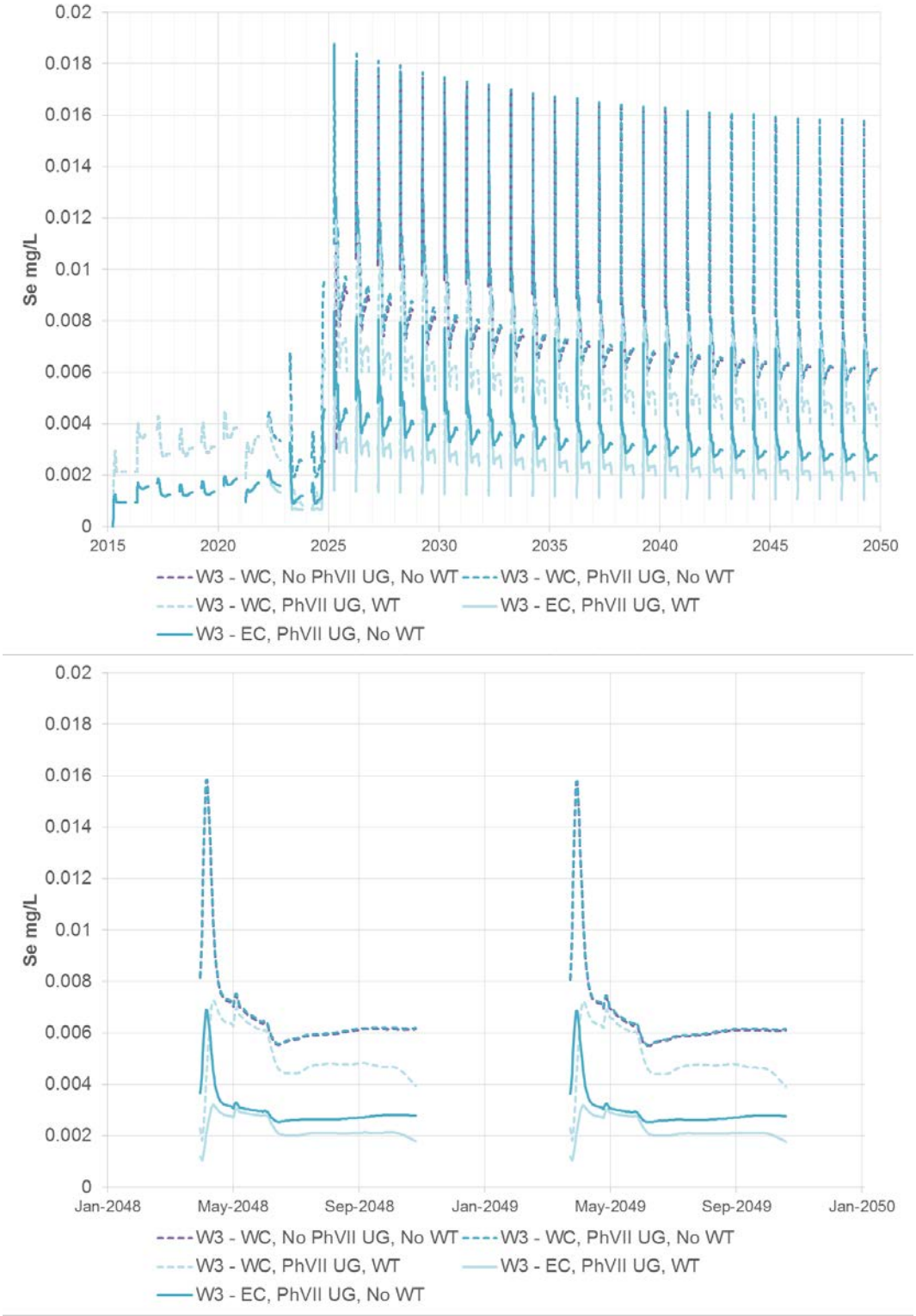
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Figure 5-6 Model Predictions of Sulphate Concentrations in Pit Lake Water



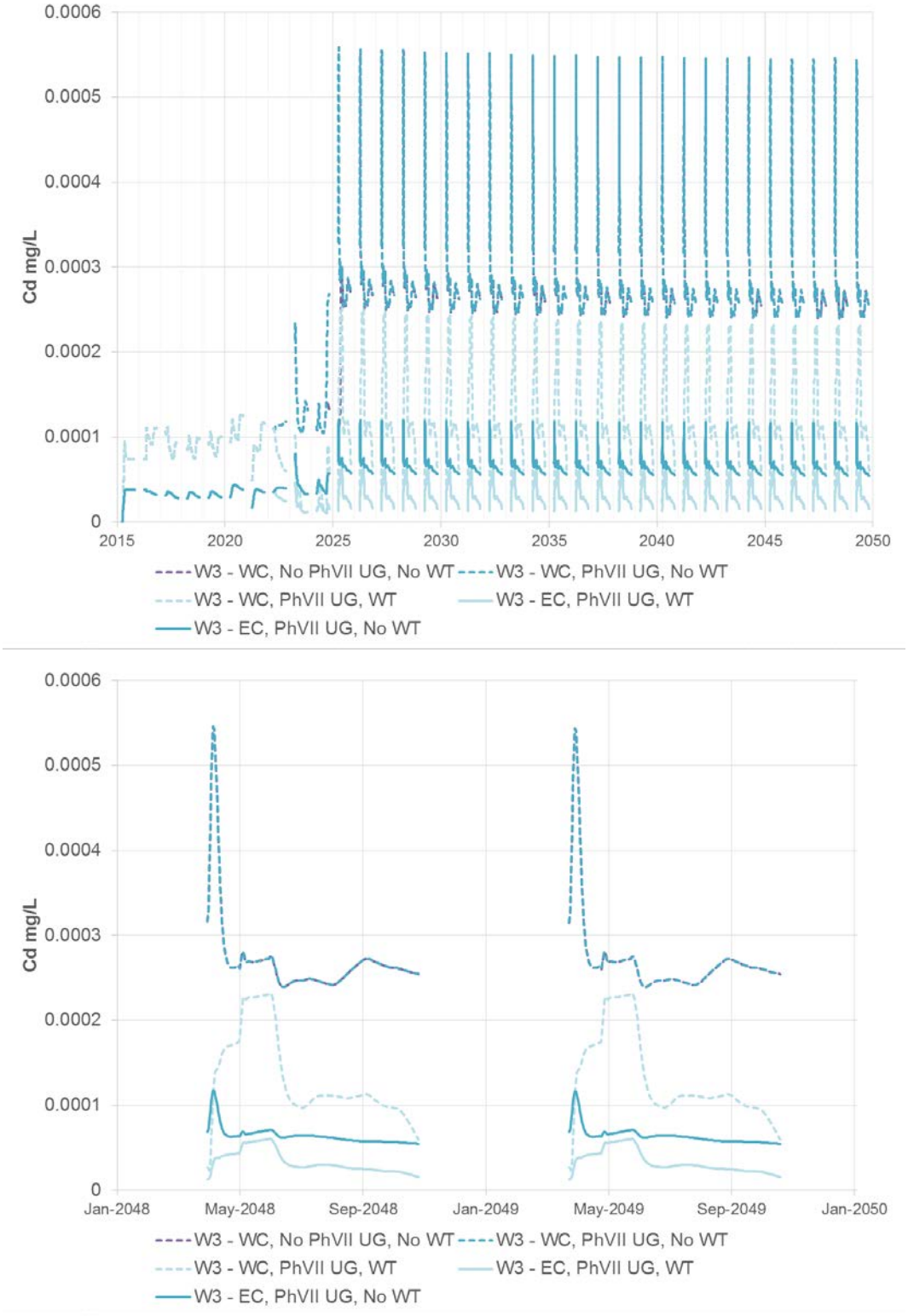
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Figure 5-7 Model Predictions of Dissolved Copper Concentrations at W3



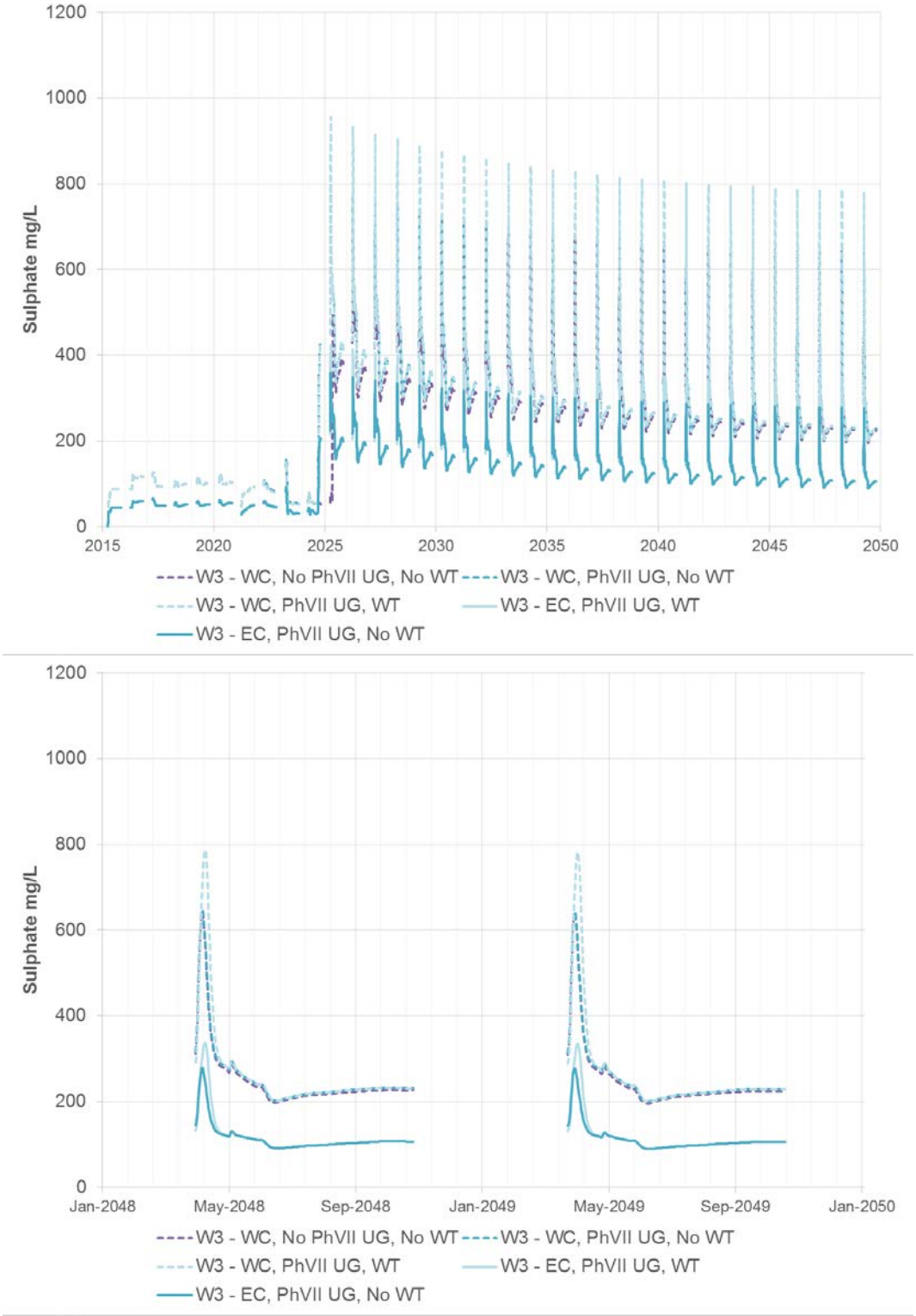
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Figure 5-8 Model Predictions of Dissolved Selenium Concentrations at W3



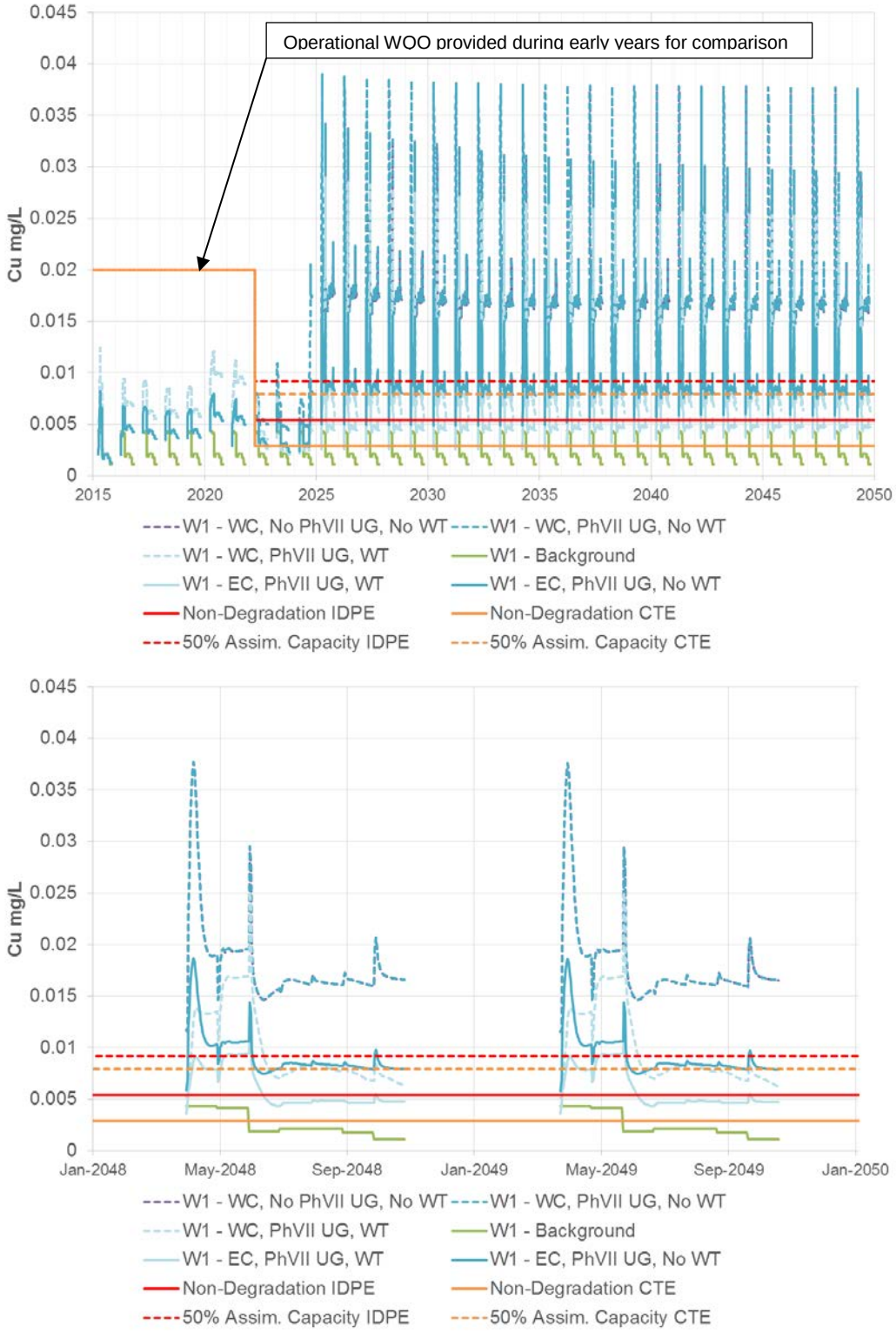
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Figure 5-9 Model Predictions of Dissolved Cadmium Concentrations at W3



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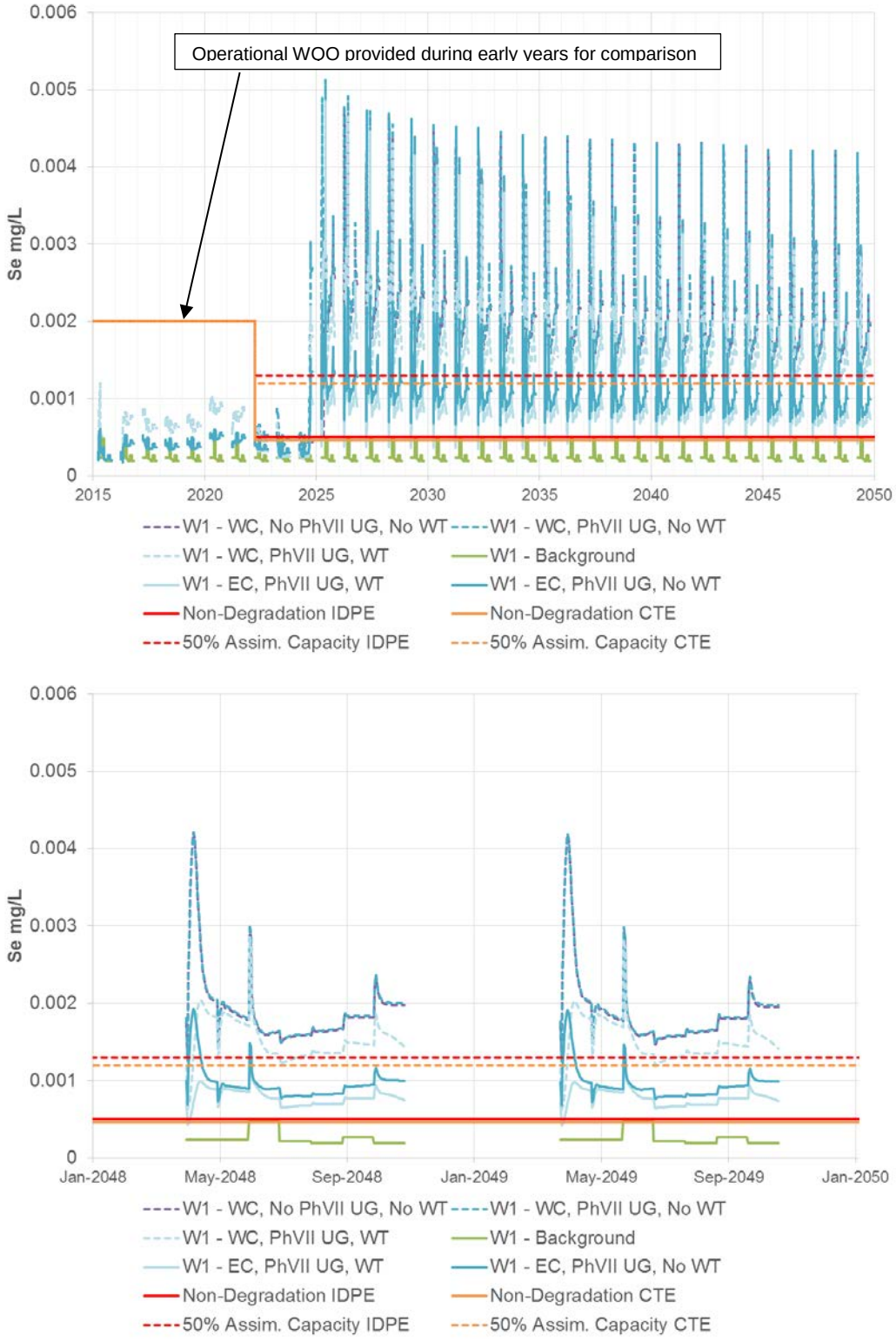
Figure 5-10 Model Predictions of Sulphate Concentrations at W3



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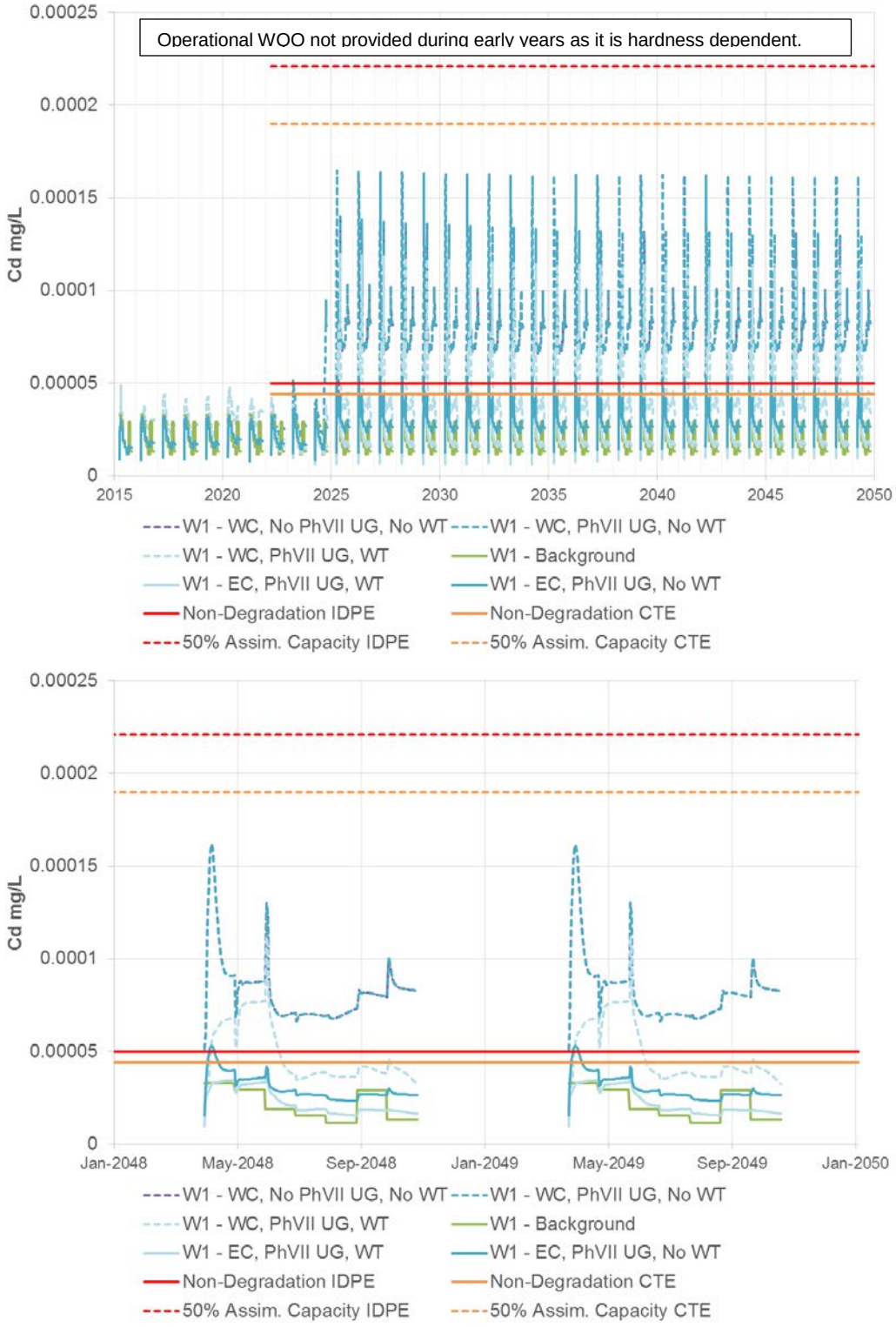
Note: The WQO for copper shown assumes dissolved organic carbon (DOC) < 10 mg/L

Figure 5-11 Model Predictions of Dissolved Copper Concentrations in Lower Minto Creek



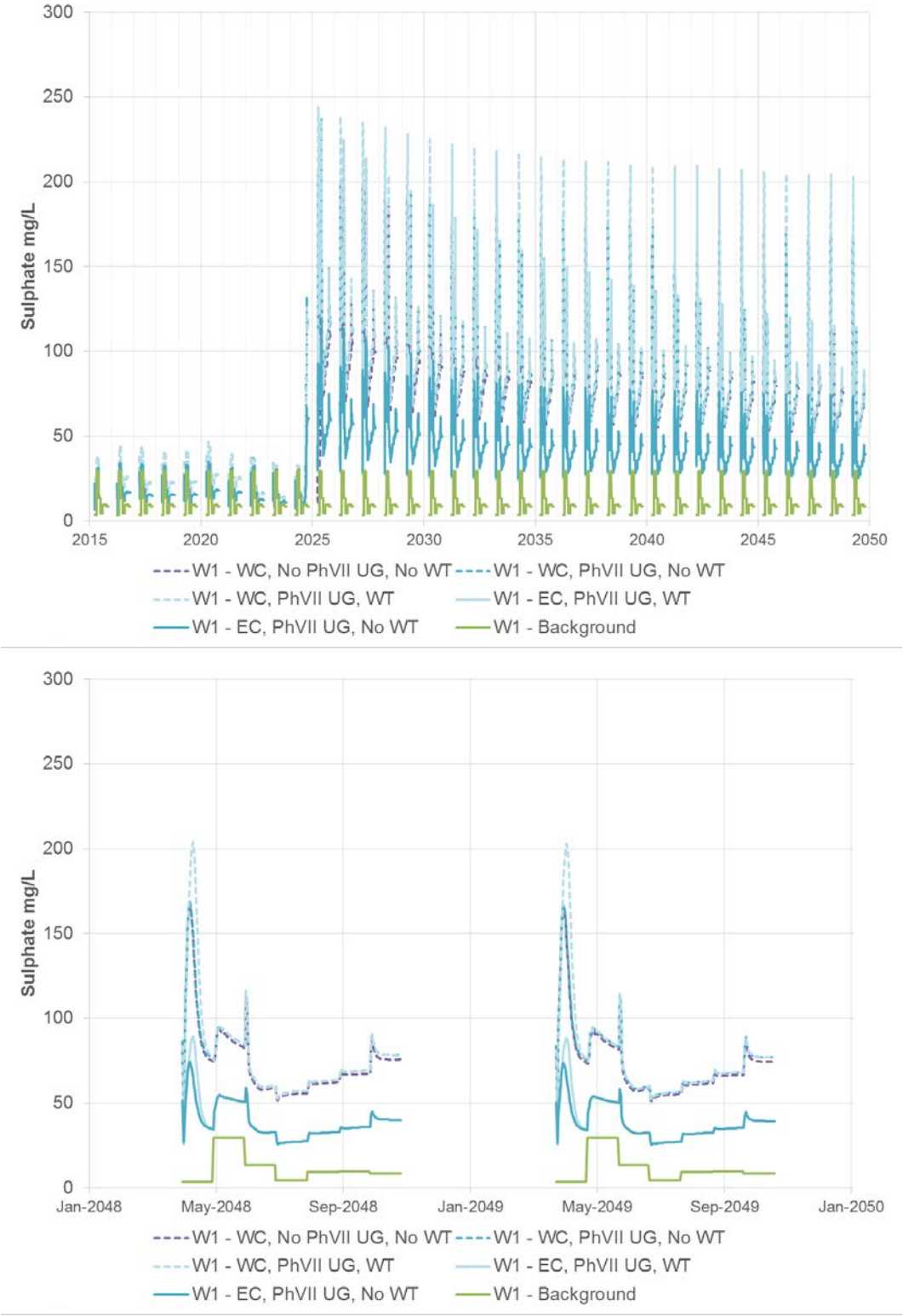
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Figure 5-12 Model Predictions of Dissolved Selenium Concentrations in Lower Minto Creek



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Figure 5-13 Model Predictions of Dissolved Cadmium Concentrations in Lower Minto Creek



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Figure 5-14 Model Predictions of Dissolved Sulphate Concentrations in Lower Minto Creek

Evaluation of the effects of the predicted water quality on aquatic resources in Minto Creek and McGinty Creek has been addressed elsewhere (Minto 2013) and is not considered here in detail. The predicted concentrations of water quality parameters include 'Expected Case' values and 'Reasonable Worst Case' values. In the context of actual project performance, the predicted 'expected case' concentrations should be considered to be typical performance values (including the range of variability observed over operations at the Minto Mine). The predictions which represent the 'Reasonable Worst Case' scenario are considered to be unlikely to occur, and should be considered to represent the upper end of the range of potential water quality performance of the mine. In the unlikely event that water quality concentrations in the range of the 'Reasonable Worst Case' values occur, it is expected that this condition would be transient and of short duration (as observed with the upper range of water quality concentrations in the site monitoring results to date).

Comparison of model results between 'no treatment' and 'passive treatment' cases shows that the CWTS has the potential to materially reduce dissolved concentrations of copper, selenium, and other parameters in surface water leaving the mine site during low flow summer and fall months when most or all of the surface water can be routed through the CWTS. However, under typical spring conditions and under high flow conditions at other times of the year, a portion of the surface water will bypass the CWTS and will therefore report to lower Minto Creek without the benefit of wetland treatment. A number of other factors will affect the performance of the CWTS, and the reclamation research into constructed wetlands treatment system that has been underway for several years will continue going forward with the intent of refining the understanding of factors that will be most critical to optimizing performance at the full scale. At present, given the range of uncertainty, it is prudent to consider the range of performance that is book-ended by the 'no treatment' and 'passive treatment' scenarios as representative of the range of future surface water quality leaving the mine site.

Comparison of model results between 'PhVII UG' and 'No PhVII UG' cases show that the inclusion of the Phase VII Underground components in the model have little to no numerical effect on the modelled water quality in the open pits and Minto Creek in post-closure. There are some very minor differences in water quality in the Main Pit and Area 2 Pit during operations but these changes are not reflected in the long-term.

5.2.3 McGinty Creek Water Quality Results

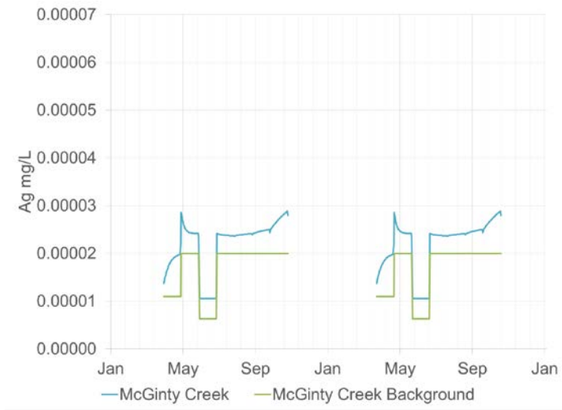
The McGinty Creek water quality modelling was a scoping evaluation for a worst-case scenario that included the following assumptions:

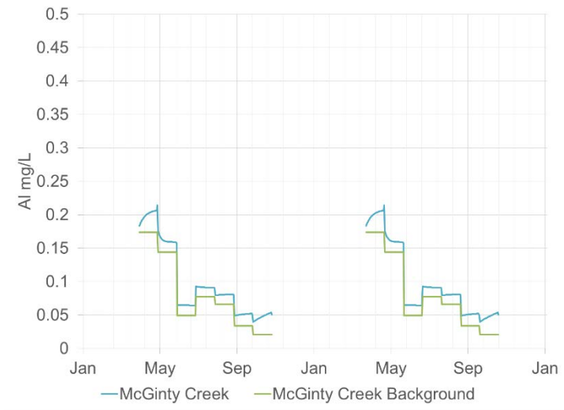
- The magnitude of loadings was set to be equal to the loads determined for the full extent of the pit walls exposed at the end of mining (i.e. there was no gradual decrease in loadings estimates even if the pit was to fill).
- All geochemical loading generated by weathering of the pit walls, pit wall failure debris, and access road is both flushed from the weathering sites and reports to water in the Minto North Pit.

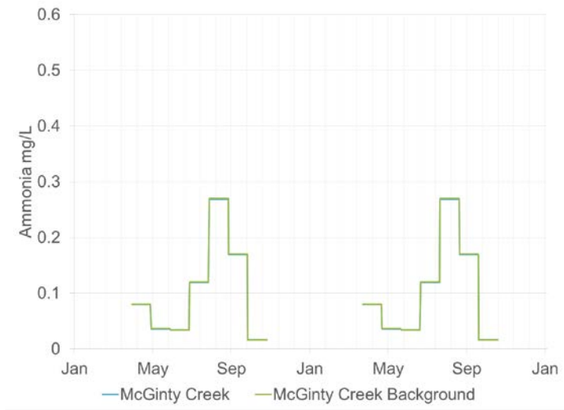
- Accumulated loadings would report unattenuated to McGinty Creek by surface discharge and/ or via groundwater pathways.

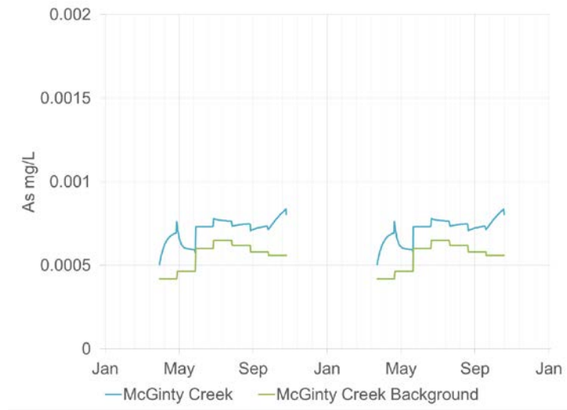
Under this scenario, Minto North loadings are maximized. This highly conservative approach evaluates the upper limit of the magnitude of potential changes to McGinty Creek water quality from geochemical loadings from the Minto North Pit.

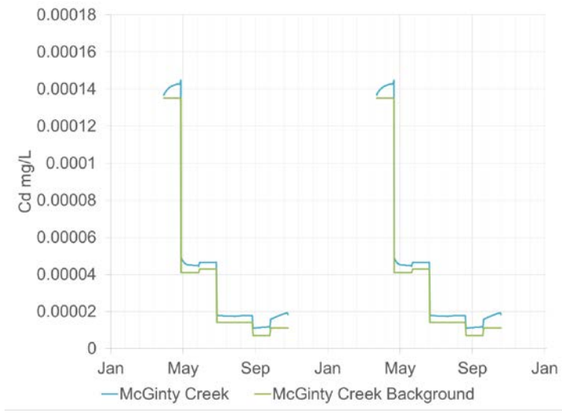
Average and maximum results of the scoping exercise for lower McGinty Creek are presented in Table 5-2, along with average and maximum values in the baseline water quality results. Long-term steady state results for selected parameters are shown in Figure 5-14a and Figure 5-15b, along with the monthly background source term representing the 95th percentile concentrations in the baseline data. Although the assessment of aquatic effects is addressed elsewhere (Minto 2013 and references therein), inspection of the baseline and worst case post-mining model results in Figure 5-14a and 5-14b reveals that the changes in water quality of McGinty Creek from mining Minto North will be minimal.

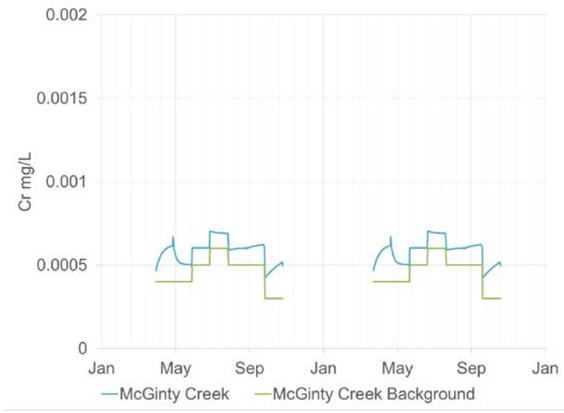


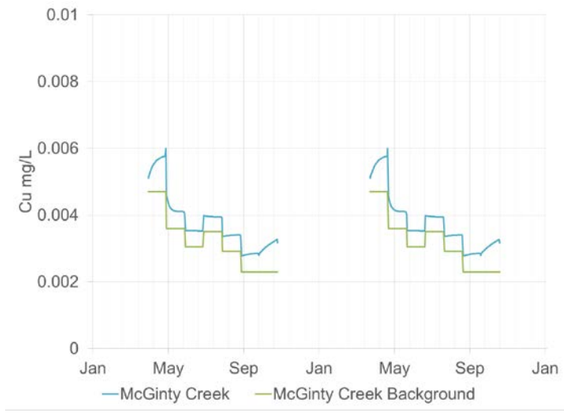












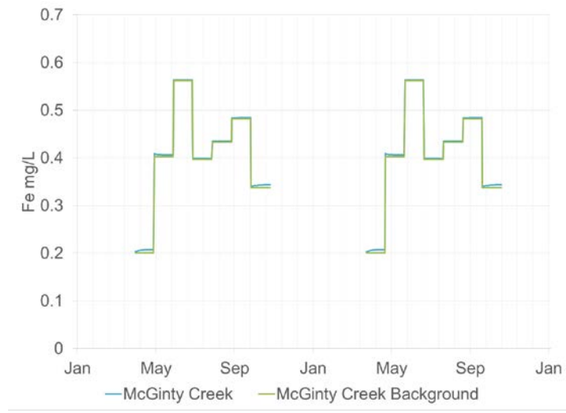
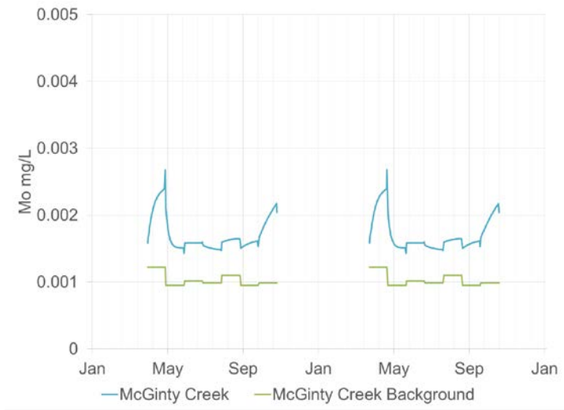
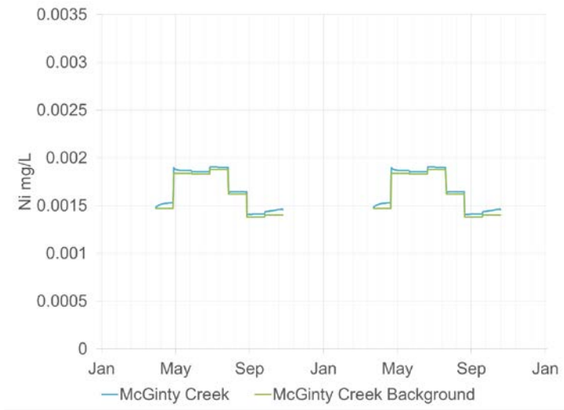
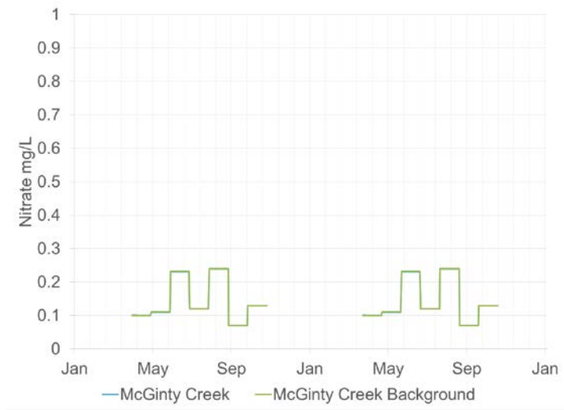
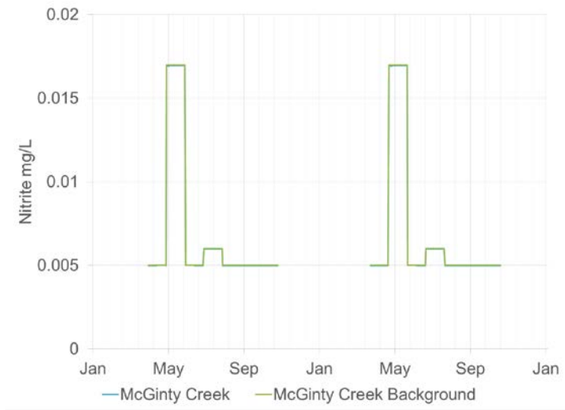


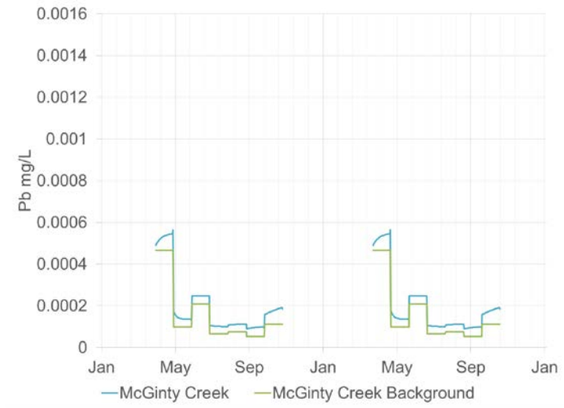
Figure 5-15a McGinty Creek Water Quality Prediction: Worst Case Results (1 of 2)

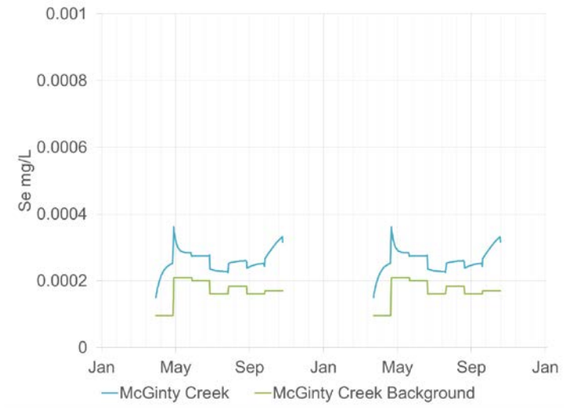


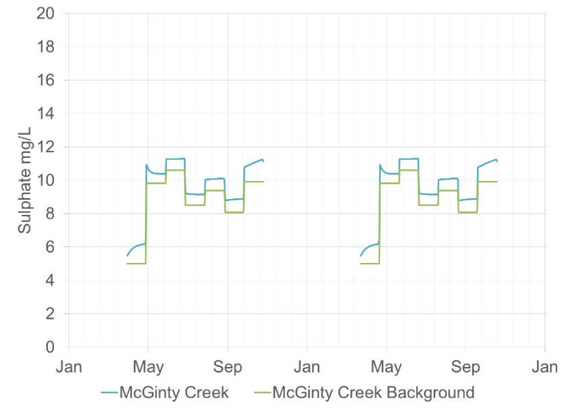












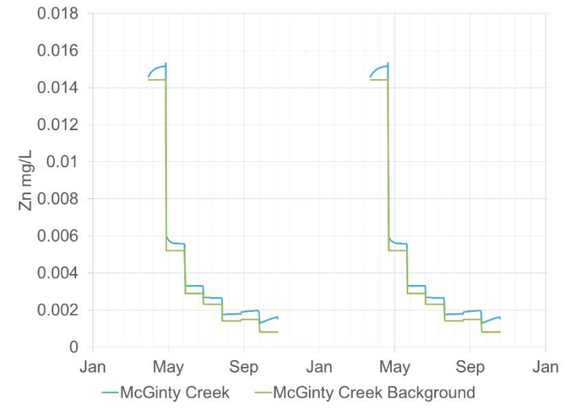


Figure 5-14b McGinty Creek Water Quality Prediction: Worst Case Results (2 of 2)

Table 5-2 Water Quality Modelling Results for McGinty Creek at Mouth

Parameter*/ Unit		Baseline Conditions		After Mining at Minto North	
				Worst Case	
		Average	Max	Average	Max
Ammonia	mg/L	0.104	0.27	0.103	0.269
$N\text{-}NO_2$	mg/L	0.00686	0.017	0.00683	0.0169
$N\text{-}NO_3$	mg/L	0.143	0.24	0.143	0.239
Sulphate	mg/L	8.75	10.6	9.55	11.3
Al	mg/L	0.0807	0.174	0.0995	0.214
As	mg/L	0.000557	0.00065	0.000715	0.000836
Cd	mg/L	0.0000378	0.000135	0.0000425	0.000145
Cr	mg/L	0.000457	0.0006	0.000583	0.000703
Cu	mg/L	0.0032	0.00471	0.00379	0.006
Fe	mg/L	0.402	0.562	0.405	0.564
Mn	mg/L	0.0275	0.0458	0.0347	0.0581
Mo	mg/L	0.00103	0.00122	0.00171	0.00268
Ni	mg/L	0.00163	0.00188	0.00167	0.00191
Se	mg/L	0.000168	0.00021	0.000262	0.000361
Ag	mg/L	0.0000168	0.00002	0.0000219	0.0000289
Zn	mg/L	0.00407	0.0144	0.00454	0.0153

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Notes:

*: all concentrations are dissolved concentrations

5.3 Limitations of the Water and Load Balance Model

The water and load balance for Minto was developed according to standard practices and was based on best available input data. However, as with any model representation of a complex system there are inherent uncertainties associated with inputs and modelled processes. In most cases, uncertainties are accounted for by using conservative assumptions that are intended to capture the most adverse conditions. One obvious exception to this approach is the water quality results from the “Expected Case” scenarios.

Uncertainties that may affect the accuracy of the model outcome include:

- Geochemical processes not incorporated in the model. The load balance is a mass conservative model. Chemical reactions such as dissolution, precipitation, oxidation or reduction are not incorporated in the model. In addition, processes such as attenuation of constituents along surface and subsurface flow paths, adsorption, and exposure of more reactive mine waste than previously-mined materials at Minto all

may contribute to actual water quality performance being different than the predictions presented here indicate.

- Significant changes to the operations and closure plans used as a basis for the modelling scenarios.
- Change in local climatic conditions that cause the historical climatic record to inaccurately represent the duration, frequency and intensity of precipitation and runoff events.

5.3.1 Variability in Precipitation

The long term results produced by the water and load balance model reflect a case where the site experiences the average annual precipitation each year, and that the annual precipitation is distributed according to a fixed monthly runoff distribution. The future reality of the site will undoubtedly deviate from that model scenario- the example shown in Figure 4-1 suggests that it is reasonable to expect at least 3-to-5-fold variability in annual precipitation over a 10-year period (i.e. the range represented by the 5th and 95th percentile results for different months). Other meteorological and hydrological parameters that affect catchment processes and stream flows in Minto Creek will undoubtedly have analogous variability (monthly distribution of runoff as just one example, as shown in Figure 2-7). For this reason, the water quality values represented by the reported model results should be considered as a range spanning the different model scenarios.

5.3.2 POX Ore and Sulphide Ore Tailings Slurry

During the Minto operations phase, the open pit water quality is sensitive to changes in the tailings slurry water quality due to the large volumes of water circulating through the mill. The tailings slurry water quality is variable over time and is linked to the amount of POX ore being milled. There is a large amount of variability in the grade of POX being milled during operations which introduces variability into the tailings source terms. The water and load balance results presented here are based on the current sulphide ore and POX ore milling schedule which is very subject to change throughout operations.

6 Key Conclusions

The Minto Mine Water and Load Balance Model was updated as part of the 2018 Reclamation and Closure Plan (Minto 2018). The key conclusions of this update are:

- that recent trends in sulphate concentrations in the in-pit tailings management facilities are likely caused by increased sulphate loading due to milling of POX ore, and
- the model results indicate that the inclusion of Phase VII mine components have very little effect on the water quality in the open pits and Minto Creek.

This report, "*Water and Load Balance Model Report 2018*", has been prepared by SRK Consulting (Canada) Inc.

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All data used as source material plus the text, tables, figures, and attachments of this document have been reviewed and prepared in accordance with generally accepted professional engineering and environmental practices.

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The opinions expressed in this report have been based on the information available to SRK at the time of preparation. SRK has exercised all due care in reviewing information supplied by others for use on this project. Whilst SRK has compared key supplied data with expected values, the accuracy of the results and conclusions from the review are entirely reliant on the accuracy and completeness of the supplied data. SRK does not accept responsibility for any errors or omissions in the supplied information, except to the extent that SRK was hired to verify the data.

7 References

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Appendix A: Water and Load Balance Results

Appendix A1: Monthly Water and Load Balance Results – 2045

Scenario 1 - Area 2 Pit 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.0000573	0.448	1.21	0.00277	0.0000819	0.00168	0.0202	0.368	0.247	0.0238	0.00222	7.27	0.356	0.00043	0.00505	207	0.00875
May	0.0000575	0.451	1.18	0.00278	0.0000823	0.00168	0.0202	0.366	0.244	0.0236	0.00221	7.12	0.349	0.000431	0.00501	204	0.00876
June	0.0000579	0.456	1.17	0.00281	0.0000829	0.00169	0.0203	0.366	0.242	0.0235	0.00221	7.04	0.346	0.000433	0.00499	202	0.0088
July	0.0000579	0.456	1.16	0.00281	0.0000829	0.00169	0.0203	0.367	0.242	0.0235	0.00222	7.03	0.345	0.000433	0.00499	202	0.0088
August	0.0000579	0.456	1.16	0.00281	0.0000828	0.00169	0.0203	0.367	0.242	0.0235	0.00222	7	0.344	0.000433	0.00498	201	0.0088
September	0.0000577	0.454	1.15	0.0028	0.0000826	0.00169	0.0202	0.367	0.241	0.0234	0.00222	6.97	0.343	0.000432	0.00496	201	0.00877
October	0.0000576	0.453	1.15	0.00279	0.0000823	0.00168	0.0201	0.367	0.24	0.0233	0.00221	6.95	0.341	0.000431	0.00495	200	0.00875

Scenario 1 - Main Pit 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.0000573	0.448	1.21	0.00277	0.0000819	0.00168	0.0202	0.368	0.247	0.0238	0.00222	7.27	0.356	0.00043	0.00505	207	0.00875
May	0.0000575	0.451	1.18	0.00278	0.0000823	0.00168	0.0202	0.366	0.244	0.0236	0.00221	7.12	0.349	0.000431	0.00501	204	0.00876
June	0.0000579	0.456	1.17	0.00281	0.0000829	0.00169	0.0203	0.366	0.242	0.0235	0.00221	7.04	0.346	0.000433	0.00499	202	0.0088
July	0.0000579	0.456	1.16	0.00281	0.0000829	0.00169	0.0203	0.367	0.242	0.0235	0.00222	7.03	0.345	0.000433	0.00499	202	0.0088
August	0.0000579	0.456	1.16	0.00281	0.0000828	0.00169	0.0203	0.367	0.242	0.0235	0.00222	7	0.344	0.000433	0.00498	201	0.0088
September	0.0000577	0.454	1.15	0.0028	0.0000826	0.00169	0.0202	0.367	0.241	0.0234	0.00222	6.97	0.343	0.000432	0.00496	201	0.00877
October	0.0000576	0.453	1.15	0.00279	0.0000823	0.00168	0.0201	0.367	0.24	0.0233	0.00221	6.95	0.341	0.000431	0.00495	200	0.00875

Scenario 1 - W3 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.0000625	0.318	0.272	0.00251	0.0000798	0.00247	0.0385	0.691	0.453	0.0141	0.00307	1.6	0.0766	0.000549	0.00448	180	0.0129
May	0.0000477	0.3	0.332	0.00198	0.0000676	0.00174	0.0297	0.535	0.35	0.0117	0.00222	1.9	0.0931	0.000407	0.00318	125	0.00935
June	0.0000479	0.304	0.253	0.00201	0.0000653	0.00174	0.0287	0.56	0.362	0.0104	0.00217	1.41	0.0702	0.0004	0.00278	102	0.00926
July	0.0000494	0.293	0.193	0.00208	0.0000639	0.00186	0.0293	0.589	0.359	0.00999	0.00232	1.06	0.053	0.000418	0.00269	98.9	0.00967
August	0.0000504	0.279	0.21	0.00208	0.00006	0.00186	0.0281	0.586	0.332	0.0102	0.00236	1.19	0.0586	0.000423	0.00272	104	0.00952
September	0.0000493	0.265	0.223	0.00203	0.0000574	0.00188	0.0268	0.577	0.32	0.0102	0.0024	1.28	0.0626	0.000428	0.00283	109	0.00933
October	0.0000472	0.255	0.237	0.00196	0.000056	0.00185	0.0254	0.553	0.305	0.0102	0.00236	1.38	0.0675	0.000422	0.00289	112	0.00906

Scenario 1 - W1 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.0000221	0.106	0.0915	0.00084	0.0000426	0.00105	0.0126	0.276	0.142	0.00426	0.00138	0.515	0.0335	0.000216	0.0013	49.7	0.00616
May	0.0000195	0.113	0.132	0.000808	0.0000351	0.000816	0.0104	0.396	0.158	0.00343	0.00151	0.525	0.0275	0.000207	0.000942	53.2	0.00554
June	0.0000174	0.0956	0.126	0.000853	0.0000302	0.000845	0.0084	0.295	0.133	0.00358	0.00151	0.514	0.0327	0.000161	0.000989	37.8	0.00517
July	0.0000182	0.0843	0.0954	0.000956	0.0000268	0.000884	0.0084	0.377	0.136	0.00324	0.00157	0.413	0.0229	0.000185	0.00082	28.3	0.00472
August	0.0000252	0.0825	0.091	0.000985	0.0000239	0.000926	0.0084	0.41	0.145	0.00333	0.00165	0.452	0.024	0.000211	0.000846	33.5	0.00466
September	0.0000198	0.0833	0.0931	0.000957	0.0000267	0.0011	0.00817	0.415	0.142	0.00341	0.00181	0.464	0.0236	0.000259	0.000955	37	0.00444
October	0.0000198	0.084	0.136	0.000934	0.0000272	0.000877	0.00825	0.363	0.154	0.00405	0.00149	0.701	0.0447	0.000187	0.00106	43	0.00413

Notes:
All concentration reported in mg/L

Scenario 2 - Area 2 Pit 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.000128	0.859	4.53	0.00689	0.000326	0.00485	0.0497	0.629	0.492	0.0646	0.00531	19.7	0.711	0.00311	0.00961	394	0.0277
May	0.000129	0.866	4.43	0.00693	0.000328	0.00488	0.0497	0.624	0.486	0.0642	0.00531	19.3	0.697	0.00313	0.00951	388	0.0277
June	0.00013	0.874	4.36	0.00699	0.000331	0.00492	0.0499	0.622	0.482	0.0643	0.00533	19	0.688	0.00316	0.00949	385	0.0279
July	0.00013	0.874	4.34	0.00699	0.000331	0.00492	0.0499	0.622	0.481	0.0643	0.00533	19	0.686	0.00316	0.00947	384	0.0279
August	0.00013	0.873	4.32	0.00698	0.00033	0.00491	0.0498	0.622	0.48	0.0642	0.00532	18.9	0.683	0.00316	0.00945	383	0.0278
September	0.00013	0.87	4.3	0.00696	0.000329	0.0049	0.0496	0.62	0.478	0.0639	0.00531	18.8	0.68	0.00315	0.00942	381	0.0277
October	0.000129	0.867	4.28	0.00694	0.000328	0.00489	0.0494	0.619	0.476	0.0637	0.0053	18.8	0.677	0.00314	0.00939	380	0.0277

Scenario 2 - Main Pit 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.0000801	0.491	0.364	0.00449	0.000309	0.00341	0.0817	1.6	1.3	0.0319	0.00664	1.53	0.0494	0.00152	0.00989	429	0.0295
May	0.000102	0.713	0.264	0.00546	0.000356	0.00418	0.0957	1.76	1.43	0.0347	0.00734	1.11	0.0379	0.00211	0.00915	354	0.0331
June	0.000113	0.776	0.208	0.006	0.000342	0.00455	0.0936	1.87	1.49	0.0348	0.00712	0.844	0.0302	0.00238	0.00792	302	0.0347
July	0.000107	0.699	0.21	0.00575	0.000311	0.00435	0.0876	1.99	1.57	0.0325	0.00709	0.855	0.0304	0.0022	0.00768	308	0.0324
August	0.000102	0.643	0.21	0.00562	0.000324	0.00425	0.0842	2.26	1.73	0.0318	0.00723	0.874	0.0303	0.00204	0.0079	317	0.031
September	0.0000961	0.59	0.214	0.00532	0.000349	0.00413	0.0803	2.22	1.63	0.0315	0.00699	0.898	0.0305	0.00188	0.00798	320	0.03
October	0.0000907	0.543	0.226	0.005	0.000326	0.00392	0.0762	2.09	1.49	0.03	0.00665	0.941	0.0318	0.00175	0.00786	324	0.0289

Scenario 2 - W3 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.000106	0.64	0.999	0.00572	0.000354	0.00439	0.0857	1.72	1.35	0.0405	0.00712	4.34	0.154	0.00214	0.0102	413	0.0324
May	0.0000878	0.591	1.2	0.00461	0.000271	0.00348	0.0647	1.09	0.876	0.034	0.00532	5.17	0.186	0.00191	0.00713	277	0.0245
June	0.0000892	0.598	0.882	0.00469	0.000253	0.00352	0.062	1.12	0.891	0.0312	0.00503	3.79	0.139	0.00195	0.00606	224	0.0243
July	0.0000895	0.579	0.656	0.00476	0.000247	0.00357	0.0634	1.28	1.01	0.0296	0.0053	2.82	0.103	0.00192	0.00601	222	0.0243
August	0.0000892	0.553	0.732	0.00476	0.000252	0.00355	0.0612	1.44	1.09	0.0299	0.0054	3.18	0.115	0.00187	0.00618	231	0.0234
September	0.0000864	0.529	0.789	0.00463	0.000269	0.00355	0.0592	1.44	1.06	0.0303	0.00534	3.44	0.124	0.0018	0.00633	237	0.0229
October	0.0000831	0.509	0.849	0.00447	0.00026	0.00347	0.0569	1.37	0.979	0.0299	0.00516	3.7	0.134	0.00176	0.00635	240	0.0223

Scenario 2 - W1 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.000033	0.187	0.318	0.00164	0.000111	0.00153	0.0243	0.53	0.366	0.011	0.00239	1.36	0.0577	0.000613	0.00275	109	0.011
May	0.0000296	0.186	0.363	0.00147	0.0000864	0.00125	0.0192	0.536	0.29	0.00909	0.00229	1.39	0.0523	0.000585	0.00194	91.9	0.00935
June	0.0000276	0.168	0.33	0.00152	0.0000767	0.00129	0.0166	0.433	0.264	0.00889	0.00222	1.27	0.0547	0.000545	0.00183	69.6	0.00891
July	0.0000277	0.151	0.243	0.00158	0.0000699	0.00129	0.0164	0.537	0.288	0.00798	0.00226	0.963	0.0389	0.000537	0.00162	58.4	0.00816
August	0.0000346	0.149	0.261	0.00163	0.0000701	0.00134	0.0164	0.614	0.328	0.00824	0.00238	1.09	0.0424	0.00056	0.0017	65.5	0.00804
September	0.0000293	0.151	0.292	0.00163	0.000081	0.00154	0.0165	0.636	0.331	0.00875	0.00257	1.2	0.0449	0.000614	0.00188	71.6	0.00797
October	0.0000303	0.158	0.446	0.00166	0.0000861	0.00135	0.0174	0.596	0.348	0.0102	0.00231	1.84	0.0777	0.000574	0.00213	84.3	0.00802

Notes:
All concentration reported in mg/L

Scenario 3 - Area 2 Pit 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.000128	0.858	3.59	0.00684	0.000325	0.00484	0.0494	0.622	0.473	0.0621	0.00524	17.4	0.609	0.0031	0.00923	357	0.027
May	0.000129	0.865	3.51	0.00688	0.000327	0.00486	0.0494	0.617	0.467	0.0619	0.00524	17.1	0.597	0.00313	0.00915	352	0.027
June	0.00013	0.873	3.46	0.00694	0.000329	0.0049	0.0496	0.615	0.464	0.062	0.00526	16.9	0.59	0.00316	0.00913	350	0.0272
July	0.00013	0.873	3.45	0.00694	0.000329	0.0049	0.0496	0.615	0.463	0.062	0.00526	16.8	0.589	0.00316	0.00912	349	0.0272
August	0.00013	0.872	3.44	0.00694	0.000329	0.0049	0.0495	0.614	0.462	0.0619	0.00526	16.8	0.587	0.00315	0.0091	348	0.0272
September	0.000129	0.87	3.42	0.00692	0.000328	0.00488	0.0493	0.613	0.46	0.0617	0.00524	16.7	0.584	0.00315	0.00907	347	0.0271
October	0.000129	0.867	3.41	0.0069	0.000327	0.00487	0.0491	0.612	0.459	0.0615	0.00523	16.6	0.582	0.00313	0.00904	346	0.027

Scenario 3 - Main Pit 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.00008	0.491	0.326	0.00448	0.000309	0.00341	0.0817	1.6	1.3	0.0318	0.00663	1.43	0.0453	0.00152	0.00988	427	0.0294
May	0.000102	0.713	0.237	0.00546	0.000356	0.00418	0.0957	1.76	1.43	0.0346	0.00734	1.04	0.035	0.00211	0.00914	353	0.0331
June	0.000113	0.776	0.188	0.006	0.000342	0.00455	0.0936	1.87	1.49	0.0347	0.00712	0.794	0.028	0.00238	0.00791	301	0.0347
July	0.000107	0.699	0.19	0.00575	0.000311	0.00435	0.0876	1.99	1.57	0.0325	0.00709	0.805	0.0282	0.0022	0.00768	307	0.0324
August	0.000102	0.643	0.19	0.00562	0.000324	0.00425	0.0842	2.26	1.73	0.0318	0.00723	0.825	0.0281	0.00204	0.0079	316	0.031
September	0.0000961	0.59	0.193	0.00532	0.000349	0.00413	0.0803	2.22	1.63	0.0315	0.00699	0.847	0.0282	0.00188	0.00797	319	0.03
October	0.0000907	0.543	0.203	0.005	0.000326	0.00392	0.0761	2.09	1.49	0.0299	0.00665	0.886	0.0293	0.00175	0.00785	323	0.0289

Scenario 3 - W3 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.000106	0.64	0.822	0.00571	0.000354	0.00439	0.0857	1.72	1.35	0.04	0.00711	3.91	0.135	0.00214	0.0101	406	0.0323
May	0.0000877	0.591	0.968	0.0046	0.000271	0.00347	0.0646	1.09	0.871	0.0334	0.0053	4.6	0.161	0.00191	0.00704	268	0.0243
June	0.0000891	0.598	0.714	0.00468	0.000252	0.00352	0.062	1.12	0.888	0.0307	0.00502	3.38	0.12	0.00195	0.00599	218	0.0241
July	0.0000895	0.579	0.536	0.00475	0.000247	0.00357	0.0634	1.28	1	0.0293	0.00529	2.52	0.0903	0.00192	0.00596	217	0.0242
August	0.0000891	0.553	0.596	0.00475	0.000252	0.00355	0.0612	1.44	1.09	0.0295	0.00539	2.85	0.1	0.00187	0.00612	225	0.0233
September	0.0000863	0.528	0.641	0.00462	0.000269	0.00355	0.0592	1.44	1.05	0.0299	0.00533	3.08	0.108	0.0018	0.00628	232	0.0228
October	0.000083	0.508	0.688	0.00447	0.000259	0.00347	0.0568	1.37	0.976	0.0295	0.00514	3.31	0.116	0.00175	0.00628	234	0.0221

Scenario 3 - W1 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.0000329	0.186	0.26	0.00164	0.000111	0.00153	0.0243	0.53	0.364	0.0108	0.00239	1.21	0.0513	0.000612	0.00273	107	0.011
May	0.0000295	0.186	0.3	0.00147	0.0000862	0.00125	0.0192	0.536	0.289	0.00893	0.00228	1.24	0.0455	0.000584	0.00192	89.5	0.00931
June	0.0000276	0.168	0.273	0.00151	0.0000766	0.00129	0.0166	0.433	0.262	0.00874	0.00222	1.14	0.0485	0.000544	0.0018	67.4	0.00887
July	0.0000276	0.151	0.203	0.00158	0.0000698	0.00129	0.0164	0.537	0.287	0.00787	0.00226	0.864	0.0344	0.000536	0.0016	56.8	0.00813
August	0.0000346	0.149	0.214	0.00163	0.0000701	0.00134	0.0164	0.614	0.327	0.00812	0.00238	0.973	0.0372	0.00056	0.00168	63.6	0.00801
September	0.0000293	0.151	0.237	0.00163	0.0000809	0.00153	0.0165	0.635	0.329	0.00861	0.00256	1.07	0.0389	0.000613	0.00186	69.4	0.00793
October	0.0000303	0.157	0.357	0.00166	0.0000859	0.00135	0.0174	0.595	0.346	0.00995	0.0023	1.62	0.068	0.000574	0.00209	80.8	0.00796

Notes:
All concentration reported in mg/L

Scenario 4 - Area 2 Pit 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.0000573	0.448	1.21	0.00277	0.0000819	0.00168	0.0202	0.368	0.247	0.0238	0.00222	7.27	0.356	0.00043	0.00505	207	0.00875
May	0.0000575	0.451	1.18	0.00278	0.0000823	0.00168	0.0202	0.366	0.244	0.0236	0.00221	7.12	0.349	0.000431	0.00501	204	0.00876
June	0.0000579	0.456	1.17	0.00281	0.0000829	0.00169	0.0203	0.366	0.242	0.0235	0.00221	7.04	0.346	0.000433	0.00499	202	0.0088
July	0.0000579	0.456	1.16	0.00281	0.0000829	0.00169	0.0203	0.367	0.242	0.0235	0.00222	7.03	0.345	0.000433	0.00499	202	0.0088
August	0.0000579	0.456	1.16	0.00281	0.0000828	0.00169	0.0203	0.367	0.242	0.0235	0.00222	7	0.344	0.000433	0.00498	201	0.0088
September	0.0000577	0.454	1.15	0.0028	0.0000826	0.00169	0.0202	0.367	0.241	0.0234	0.00222	6.97	0.343	0.000432	0.00496	201	0.00877
October	0.0000576	0.453	1.15	0.00279	0.0000823	0.00168	0.0201	0.367	0.24	0.0233	0.00221	6.95	0.341	0.000431	0.00495	200	0.00875

Scenario 4 - Main Pit 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.0000517	0.234	0.0966	0.00197	0.0000608	0.00215	0.0373	0.596	0.422	0.0112	0.00266	0.533	0.0217	0.000466	0.00401	170	0.0111
May	0.0000615	0.344	0.0756	0.00235	0.0000793	0.0024	0.0458	0.759	0.543	0.0113	0.00277	0.388	0.0172	0.000534	0.00372	138	0.0124
June	0.0000661	0.383	0.0665	0.00257	0.0000824	0.00251	0.0452	0.859	0.586	0.0112	0.00283	0.3	0.0139	0.000564	0.00336	121	0.0129
July	0.0000641	0.346	0.0679	0.00248	0.0000756	0.00249	0.0421	0.815	0.525	0.011	0.00288	0.304	0.0144	0.000553	0.00322	126	0.0129
August	0.0000627	0.314	0.0673	0.00241	0.0000694	0.00249	0.0402	0.788	0.478	0.0108	0.00292	0.312	0.0143	0.000546	0.00322	132	0.0127
September	0.0000606	0.285	0.0667	0.0023	0.0000638	0.00243	0.0377	0.762	0.451	0.0104	0.0029	0.324	0.0143	0.000533	0.0033	135	0.0124
October	0.0000579	0.263	0.0686	0.00218	0.0000597	0.00237	0.0351	0.726	0.423	0.0101	0.00284	0.337	0.0147	0.000517	0.00332	137	0.012

Scenario 4 - W3 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.000072	0.356	0.273	0.00287	0.0000357	0.00287	0.0189	0.809	0.528	0.0142	0.00356	1.6	0.0754	0.000636	0.0026	207	0.00506
May	0.0000477	0.299	0.332	0.00198	0.0000556	0.00174	0.0241	0.533	0.349	0.0115	0.00222	1.9	0.093	0.000406	0.00297	125	0.00771
June	0.000048	0.305	0.258	0.00202	0.0000411	0.00174	0.0184	0.56	0.362	0.0102	0.00217	1.44	0.0717	0.0004	0.0024	103	0.00582
July	0.0000494	0.293	0.192	0.00208	0.0000284	0.00186	0.013	0.589	0.36	0.00958	0.00232	1.05	0.0527	0.000417	0.00212	98.7	0.00438
August	0.0000504	0.279	0.21	0.00208	0.0000273	0.00186	0.013	0.586	0.333	0.0098	0.00236	1.19	0.0585	0.000423	0.00216	104	0.00442
September	0.0000494	0.265	0.222	0.00203	0.0000239	0.00188	0.013	0.578	0.32	0.00976	0.0024	1.28	0.0623	0.000428	0.00218	109	0.00392
October	0.0000472	0.256	0.238	0.00197	0.0000199	0.00185	0.0131	0.554	0.305	0.00965	0.00236	1.38	0.0677	0.000423	0.00209	112	0.00315

Scenario 4 - W1 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.0000246	0.116	0.0917	0.000934	0.0000319	0.00116	0.00778	0.306	0.161	0.0043	0.00151	0.515	0.0332	0.000238	0.000845	56.6	0.00427
May	0.0000195	0.113	0.132	0.000808	0.0000322	0.000815	0.00904	0.395	0.158	0.0034	0.00151	0.525	0.0275	0.000206	0.000889	53.3	0.00513
June	0.0000174	0.0958	0.127	0.000854	0.0000248	0.000845	0.00607	0.295	0.133	0.00354	0.00151	0.52	0.033	0.000161	0.000904	37.9	0.0044
July	0.0000182	0.0844	0.0952	0.000956	0.0000186	0.000884	0.00463	0.377	0.137	0.00315	0.00157	0.411	0.0228	0.000185	0.000688	28.3	0.0035
August	0.0000252	0.0827	0.0909	0.000985	0.0000162	0.000926	0.00482	0.41	0.145	0.00325	0.00165	0.451	0.024	0.000211	0.000714	33.5	0.00346
September	0.0000198	0.0835	0.0929	0.000958	0.0000183	0.0011	0.0047	0.415	0.142	0.00331	0.00181	0.462	0.0235	0.000259	0.000794	36.9	0.00309
October	0.0000199	0.0842	0.136	0.000935	0.0000175	0.000878	0.00488	0.363	0.154	0.00391	0.00149	0.702	0.0447	0.000187	0.000852	43	0.00254

Notes:
All concentration reported in mg/L

Scenario 5 - Area 2 Pit 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.000128	0.859	4.53	0.00689	0.000326	0.00485	0.0497	0.629	0.492	0.0646	0.00531	19.7	0.711	0.00311	0.00961	394	0.0277
May	0.000129	0.866	4.43	0.00693	0.000328	0.00488	0.0497	0.624	0.486	0.0642	0.00531	19.3	0.697	0.00313	0.00951	388	0.0277
June	0.00013	0.874	4.36	0.00699	0.000331	0.00492	0.0499	0.622	0.482	0.0643	0.00533	19	0.688	0.00316	0.00949	385	0.0279
July	0.00013	0.874	4.34	0.00699	0.000331	0.00492	0.0499	0.622	0.481	0.0643	0.00533	19	0.686	0.00316	0.00947	384	0.0279
August	0.00013	0.873	4.32	0.00698	0.00033	0.00491	0.0498	0.622	0.48	0.0642	0.00532	18.9	0.683	0.00316	0.00945	383	0.0278
September	0.00013	0.87	4.3	0.00696	0.000329	0.0049	0.0496	0.62	0.478	0.0639	0.00531	18.8	0.68	0.00315	0.00942	381	0.0277
October	0.000129	0.867	4.28	0.00694	0.000328	0.00489	0.0494	0.619	0.476	0.0637	0.0053	18.8	0.677	0.00314	0.00939	380	0.0277

Scenario 5 - Main Pit 2045 Monthly Water Quality Results

	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.0000801	0.491	0.364	0.00449	0.000309	0.00341	0.0817	1.6	1.3	0.0319	0.00664	1.53	0.0494	0.00152	0.00989	429	0.0295
May	0.000102	0.713	0.264	0.00546	0.000356	0.00418	0.0957	1.76	1.43	0.0347	0.00734	1.11	0.0379	0.00211	0.00915	354	0.0331
June	0.000113	0.776	0.208	0.006	0.000342	0.00455	0.0936	1.87	1.49	0.0348	0.00712	0.844	0.0302	0.00238	0.00792	302	0.0347
July	0.000107	0.699	0.21	0.00575	0.000311	0.00435	0.0876	1.99	1.57	0.0325	0.00709	0.855	0.0304	0.0022	0.00768	308	0.0324
August	0.000102	0.643	0.21	0.00562	0.000324	0.00425	0.0842	2.26	1.73	0.0318	0.00723	0.874	0.0303	0.00204	0.0079	317	0.031
September	0.0000961	0.59	0.214	0.00532	0.000349	0.00413	0.0803	2.22	1.63	0.0315	0.00699	0.898	0.0305	0.00188	0.00798	320	0.03
October	0.0000907	0.543	0.226	0.005	0.000326	0.00392	0.0762	2.09	1.49	0.03	0.00665	0.941	0.0318	0.00175	0.00786	324	0.0289

Scenario 5 - W3 2045 Monthly Water Quality Results

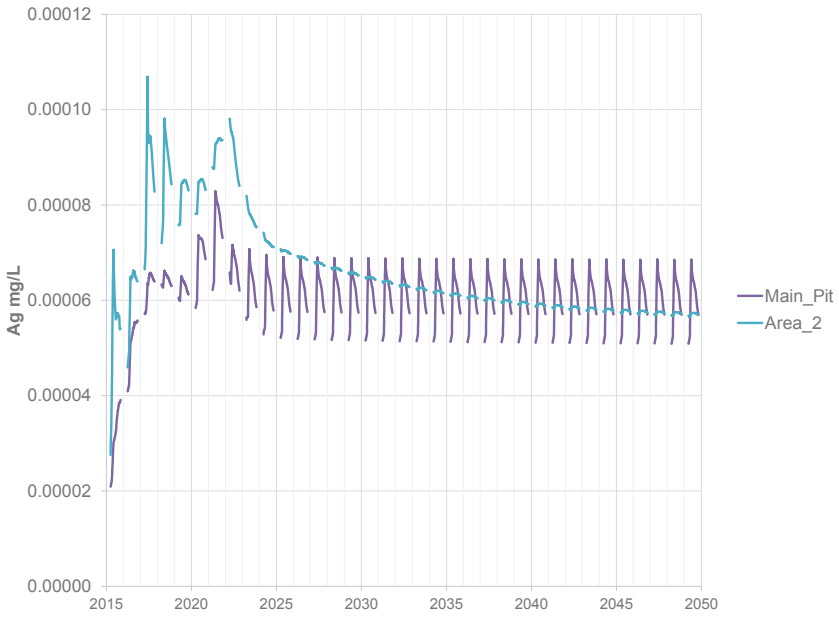
	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.000121	0.72	0.997	0.00653	0.000136	0.00502	0.0355	2.05	1.61	0.0438	0.0082	4.34	0.153	0.00242	0.00574	475	0.0123
May	0.0000876	0.59	1.2	0.0046	0.000221	0.00347	0.0524	1.09	0.875	0.0338	0.00532	5.17	0.186	0.0019	0.00665	278	0.02
June	0.0000894	0.6	0.902	0.0047	0.000156	0.00353	0.0372	1.12	0.89	0.0311	0.00503	3.87	0.142	0.00195	0.00522	225	0.015
July	0.0000895	0.58	0.652	0.00475	0.000105	0.00357	0.0262	1.27	1	0.0292	0.00529	2.8	0.103	0.00192	0.00472	222	0.0105
August	0.0000893	0.554	0.731	0.00476	0.00011	0.00356	0.0263	1.44	1.09	0.0295	0.0054	3.18	0.115	0.00187	0.0049	231	0.0104
September	0.0000865	0.529	0.785	0.00464	0.000107	0.00356	0.0229	1.44	1.06	0.0299	0.00535	3.43	0.123	0.00181	0.00488	237	0.00926
October	0.0000833	0.51	0.851	0.00448	0.0000851	0.00348	0.0191	1.37	0.982	0.0295	0.00517	3.72	0.134	0.00176	0.00457	240	0.00739

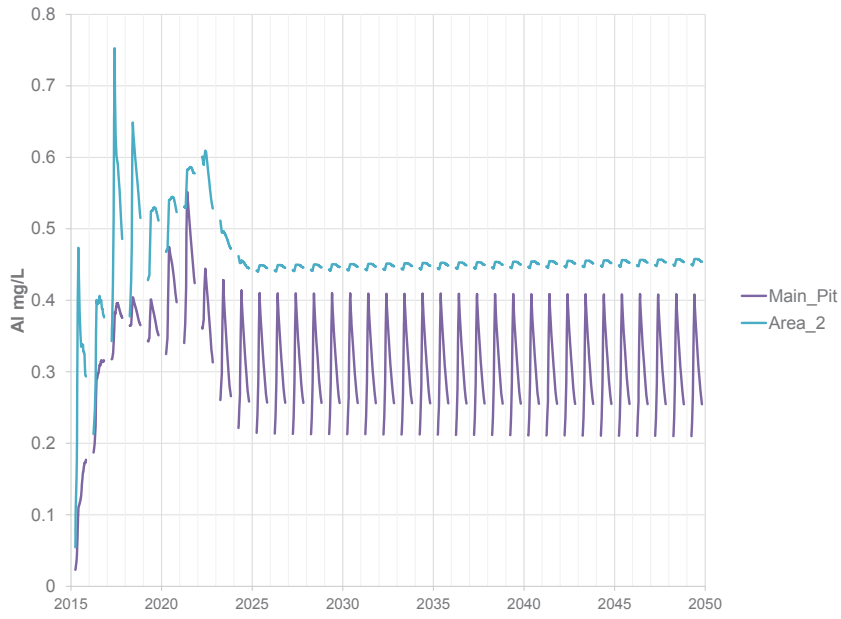
Scenario 5 - W1 2045 Monthly Water Quality Results

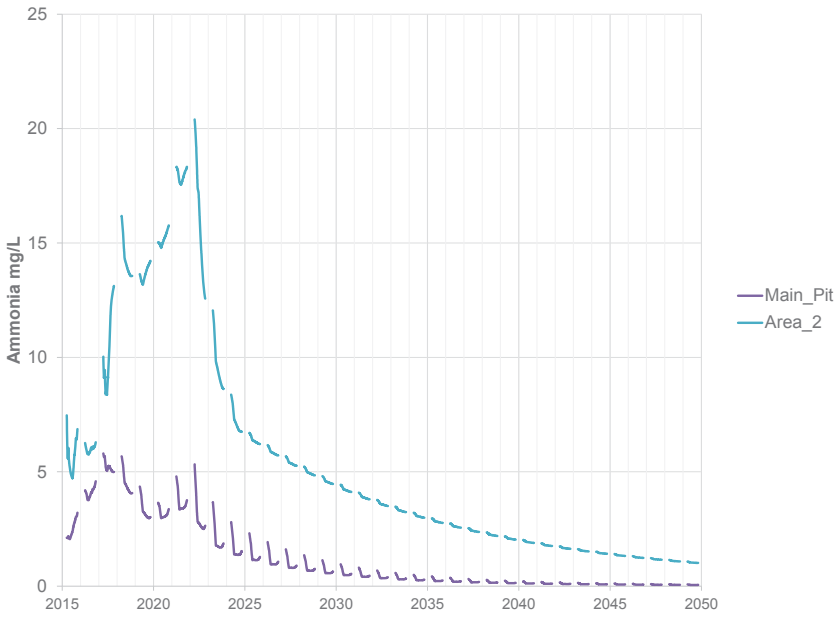
	Ag-Dissolved	Al-Dissolved	Ammonia	As-Dissolved	Cd-Dissolved	Cr-Dissolved	Cu-Dissolved	Fe-Dissolved	Mn-Dissolved	Mo-Dissolved	Ni-Dissolved	Nitrate	Nitrite	Pb-Dissolved	Se-Dissolved	Sulphate	Zn-Dissolved
April	0.0000368	0.207	0.318	0.00185	0.0000577	0.0017	0.0121	0.616	0.431	0.0118	0.00267	1.36	0.0573	0.000683	0.00166	125	0.00616
May	0.0000295	0.186	0.363	0.00147	0.0000738	0.00125	0.0162	0.536	0.29	0.00906	0.00229	1.39	0.0523	0.000584	0.00182	92.1	0.00824
June	0.0000276	0.169	0.334	0.00152	0.000055	0.00129	0.0111	0.433	0.263	0.00887	0.00222	1.29	0.0553	0.000546	0.00164	69.9	0.00683
July	0.0000277	0.151	0.242	0.00158	0.0000372	0.00129	0.00783	0.536	0.287	0.00788	0.00226	0.959	0.0387	0.000536	0.00132	58.3	0.00499
August	0.0000346	0.149	0.261	0.00163	0.0000368	0.00134	0.00816	0.614	0.328	0.00815	0.00238	1.09	0.0423	0.000561	0.0014	65.4	0.00498
September	0.0000293	0.152	0.291	0.00163	0.0000404	0.00154	0.00742	0.636	0.331	0.00865	0.00257	1.2	0.0448	0.000614	0.00152	71.6	0.00455
October	0.0000304	0.158	0.446	0.00167	0.0000391	0.00135	0.00715	0.596	0.349	0.0101	0.00231	1.84	0.0777	0.000575	0.00166	84.3	0.00403

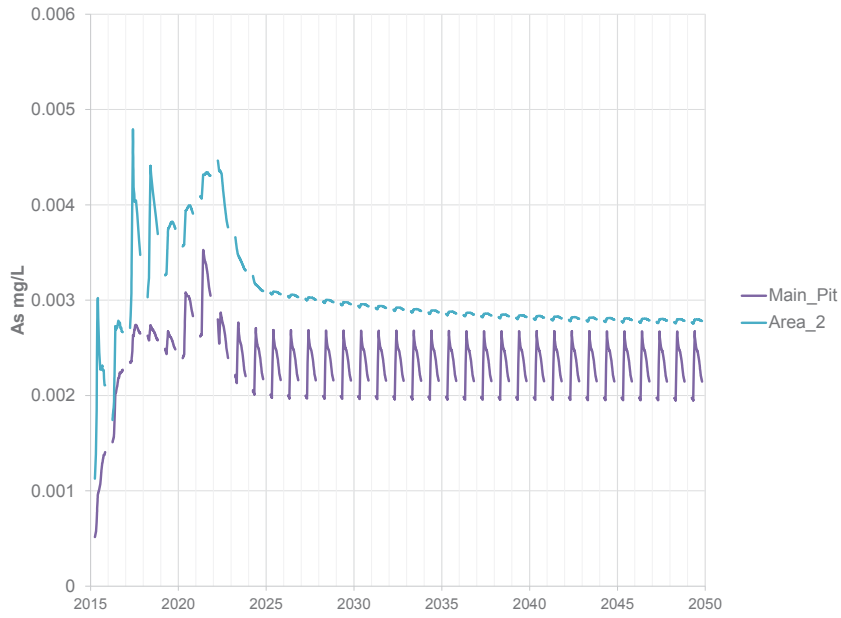
Notes:
All concentration reported in mg/L

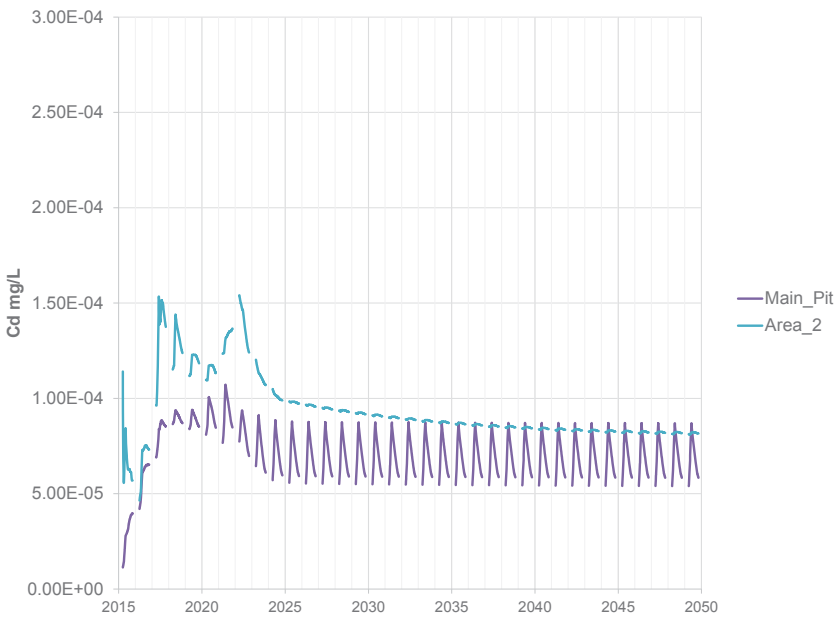
Appendix A2: Water and Load Balance Results Plots

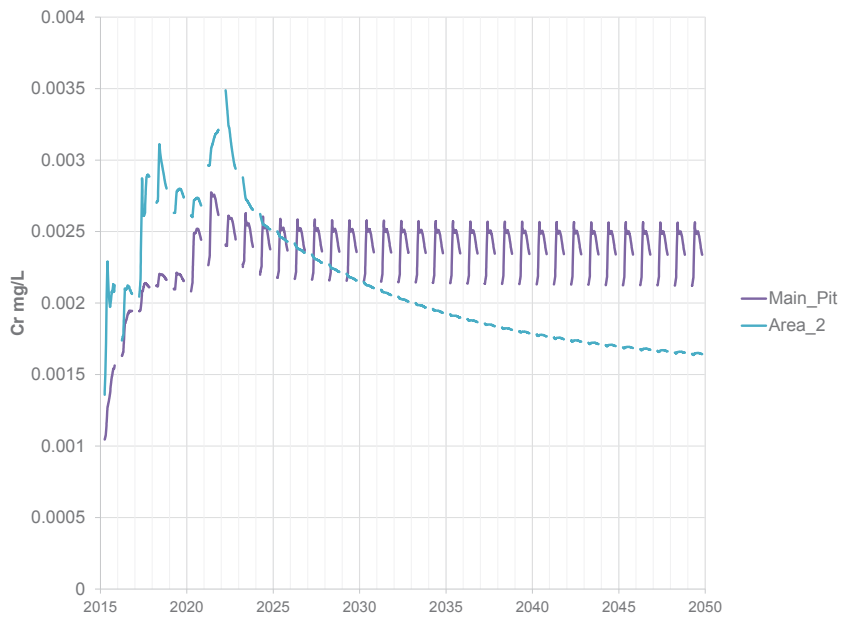


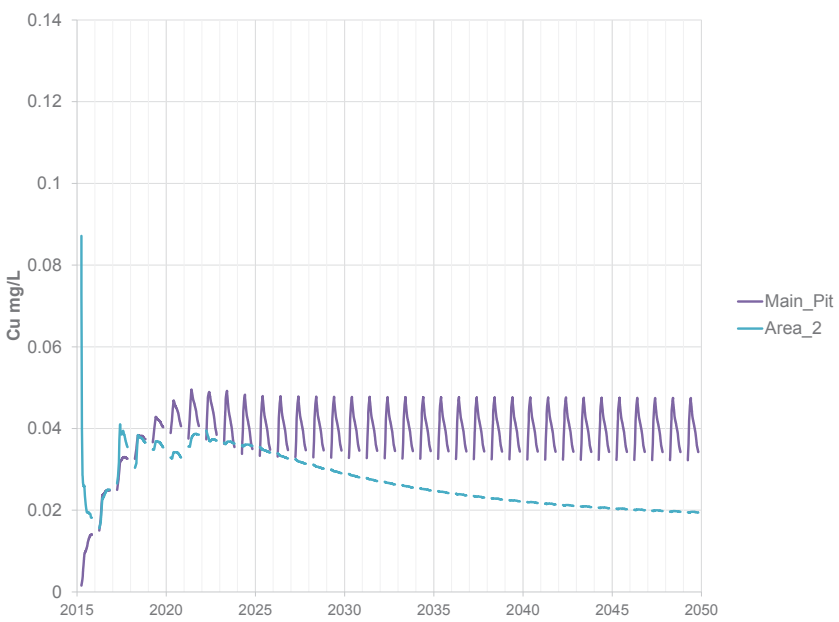


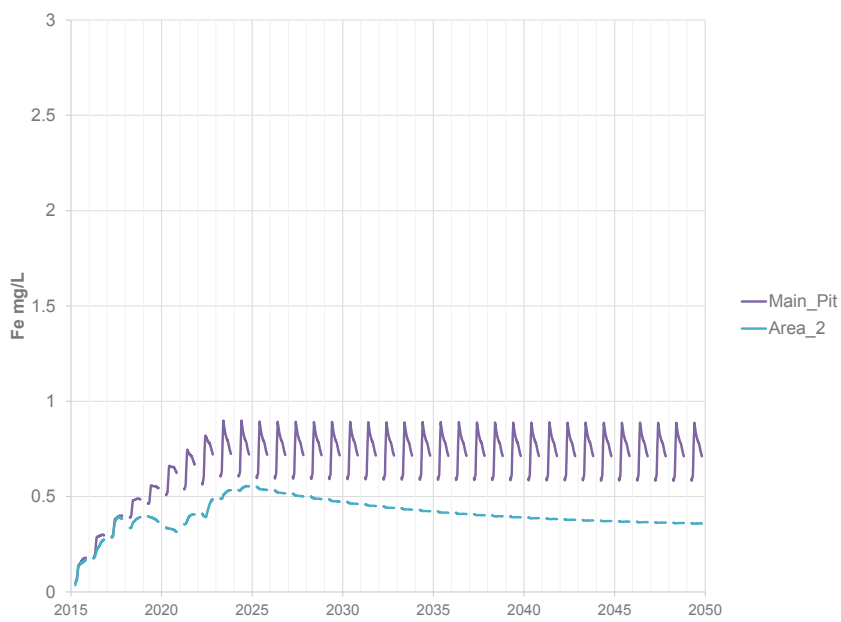




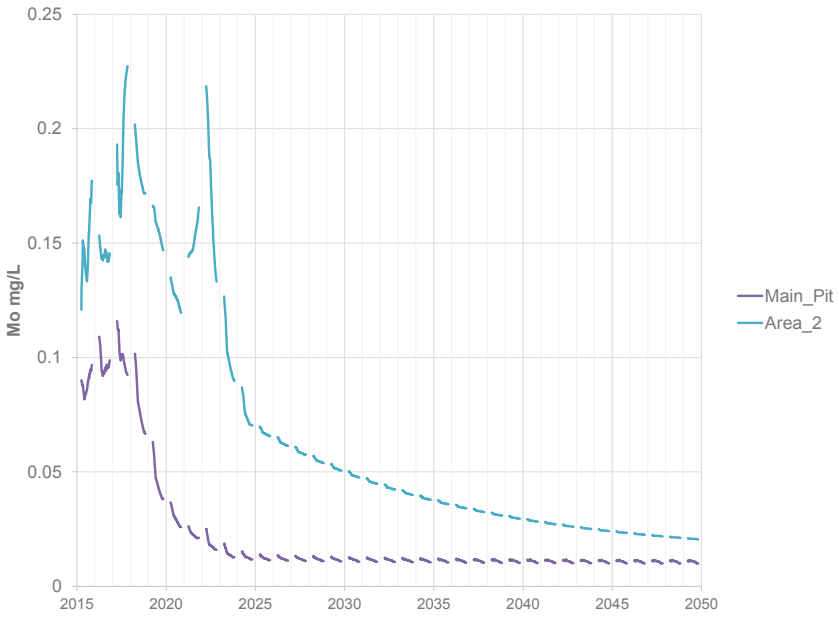


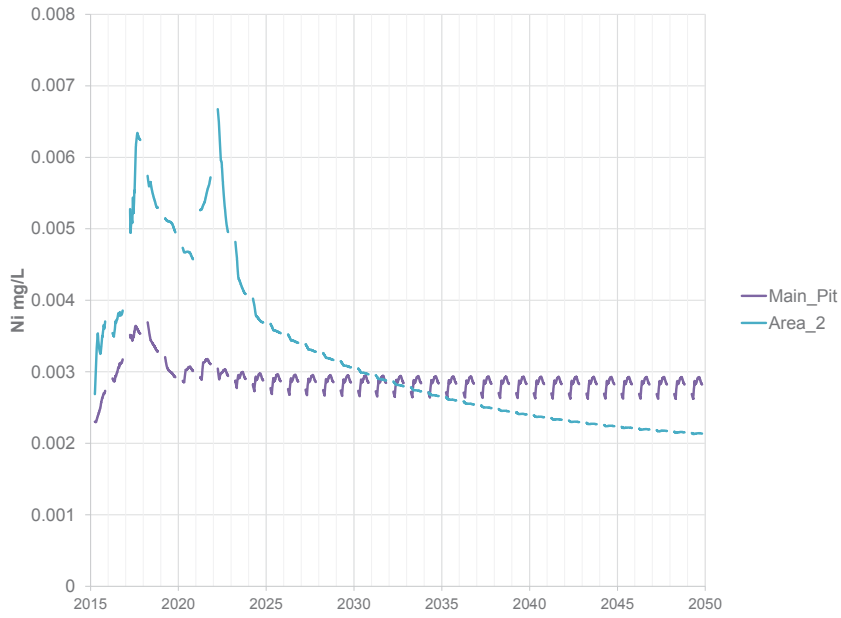


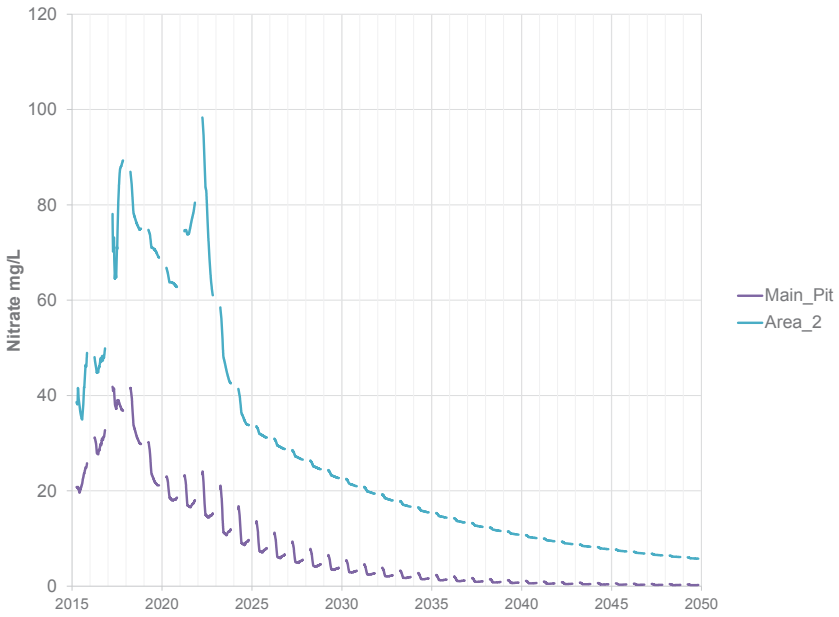


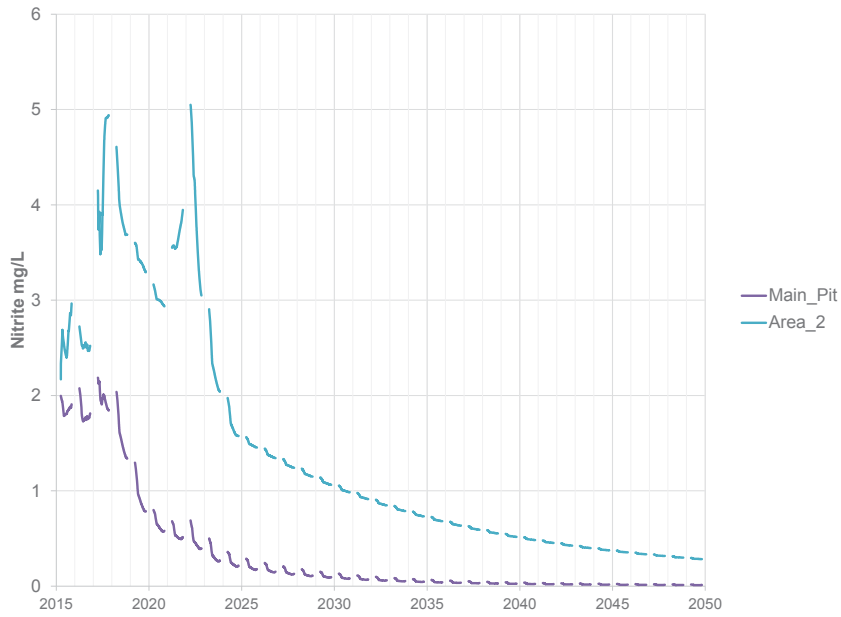


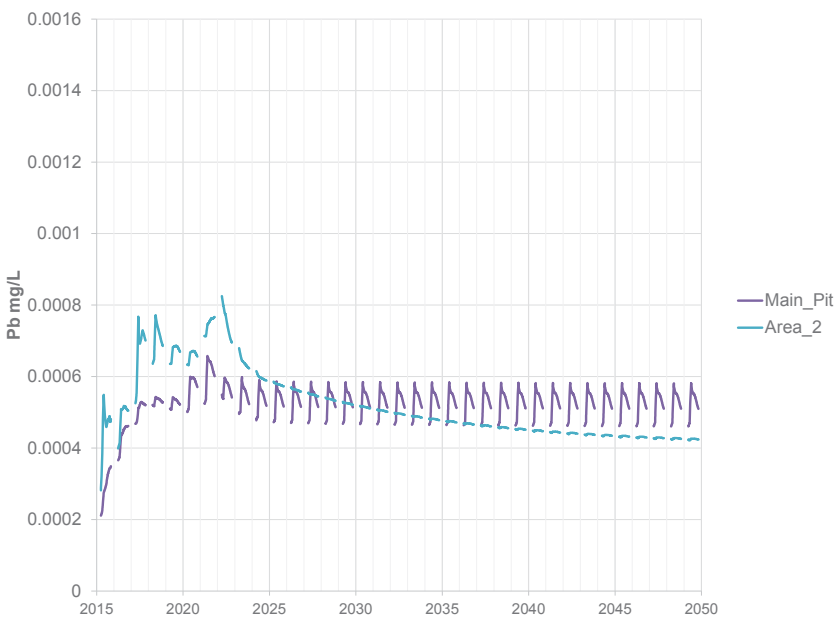
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Model: Minto_2017_1CM002.024_Rev71_SRJ_KNK, 2018/02/06, KNK

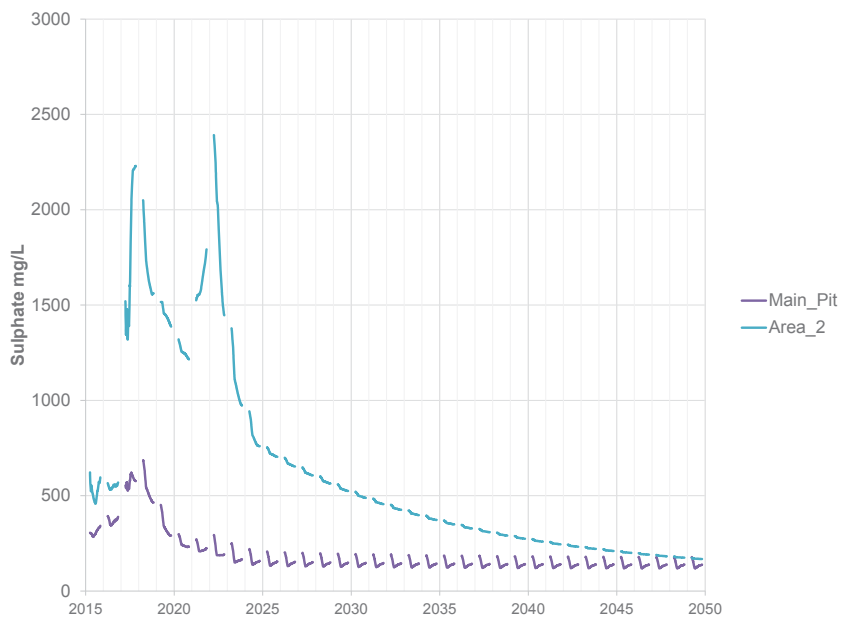


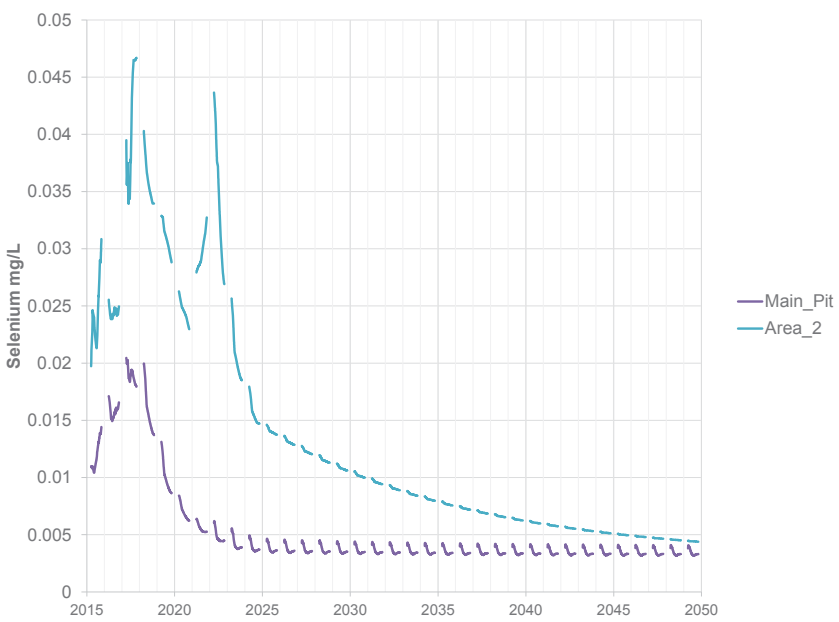


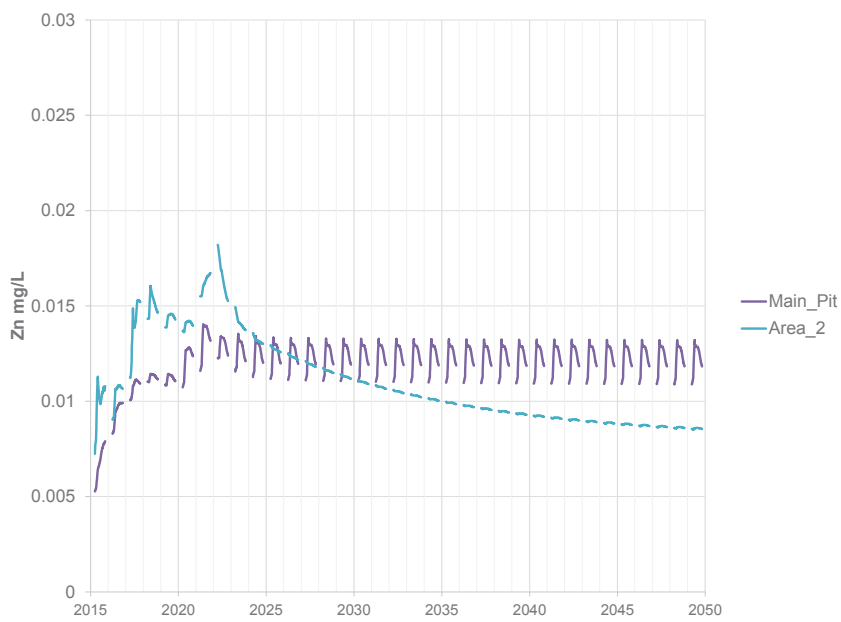




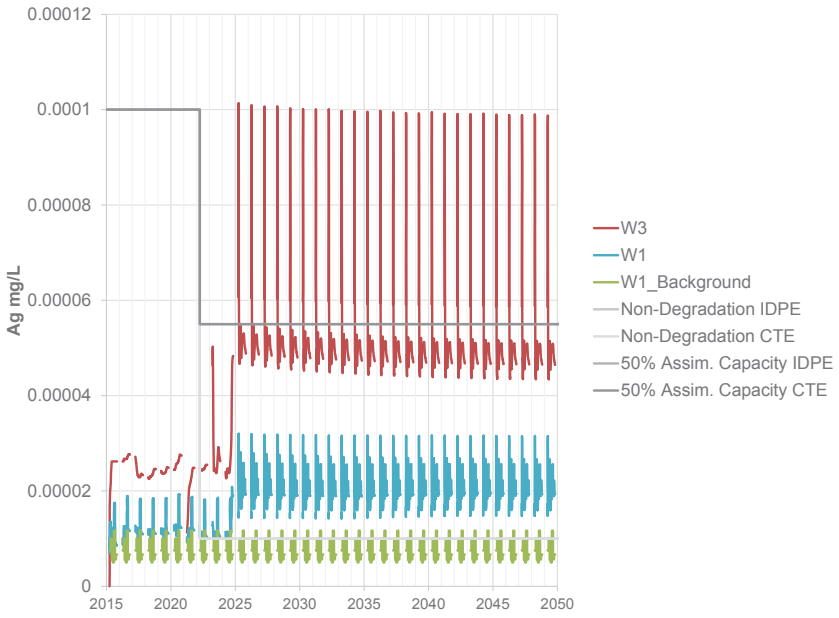


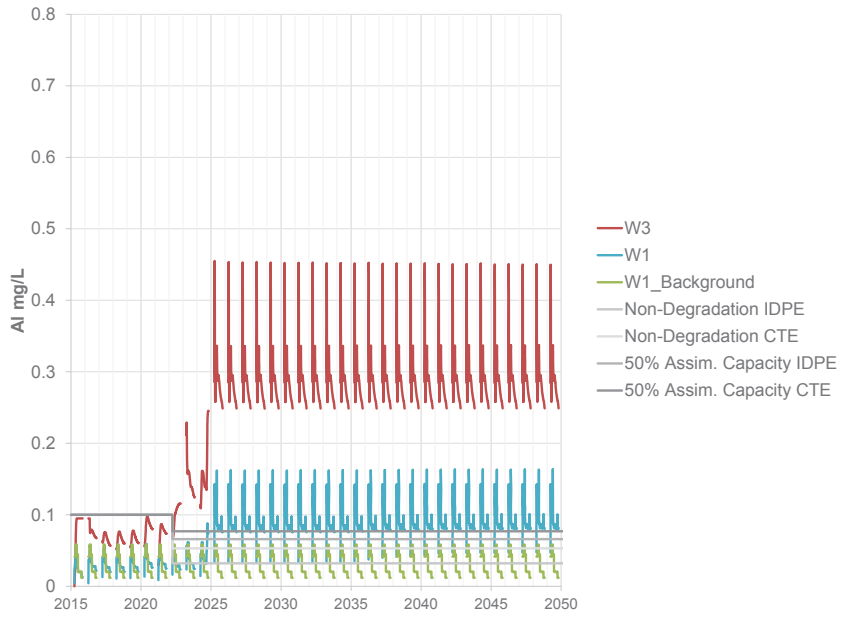


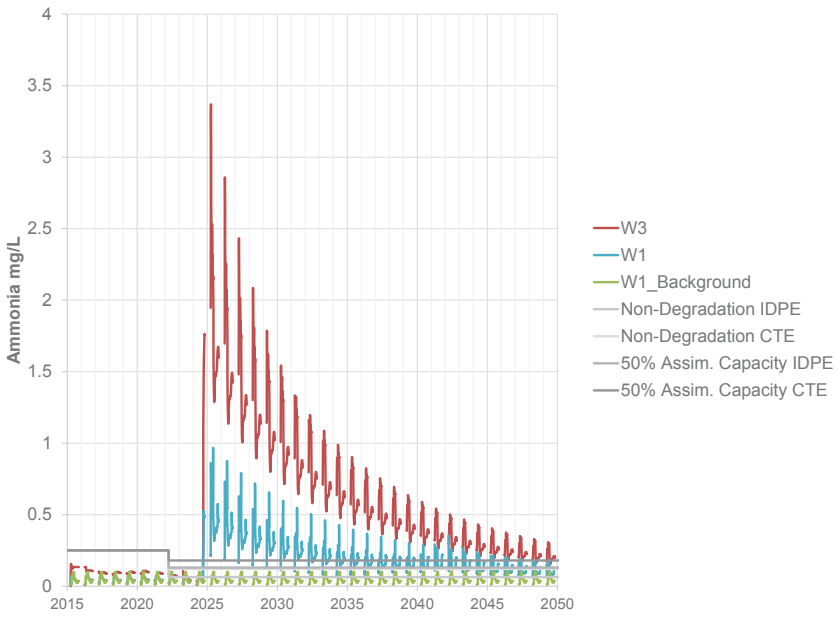


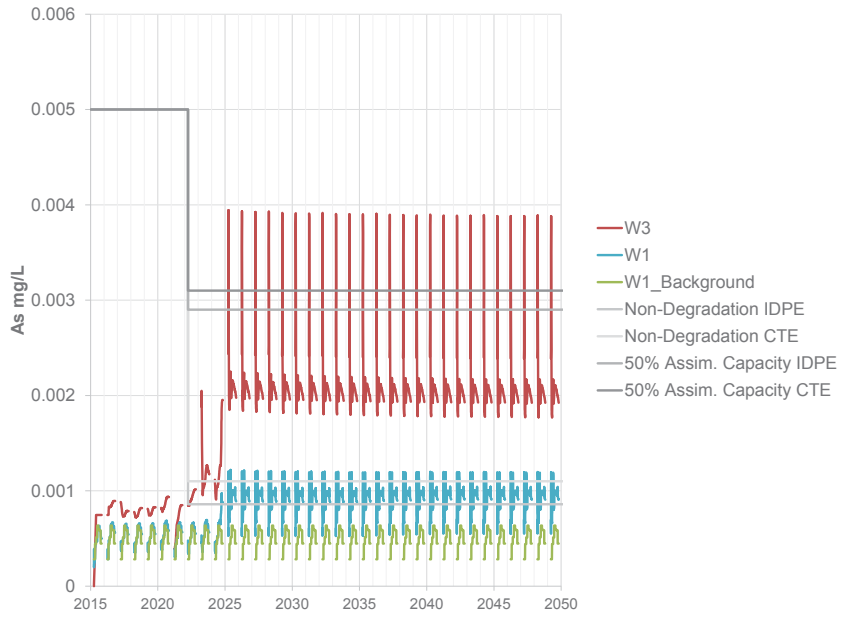


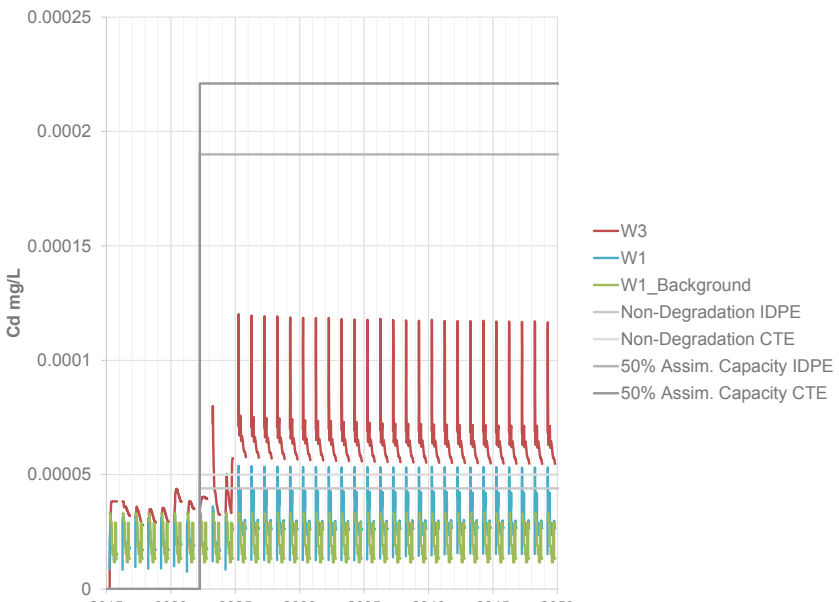
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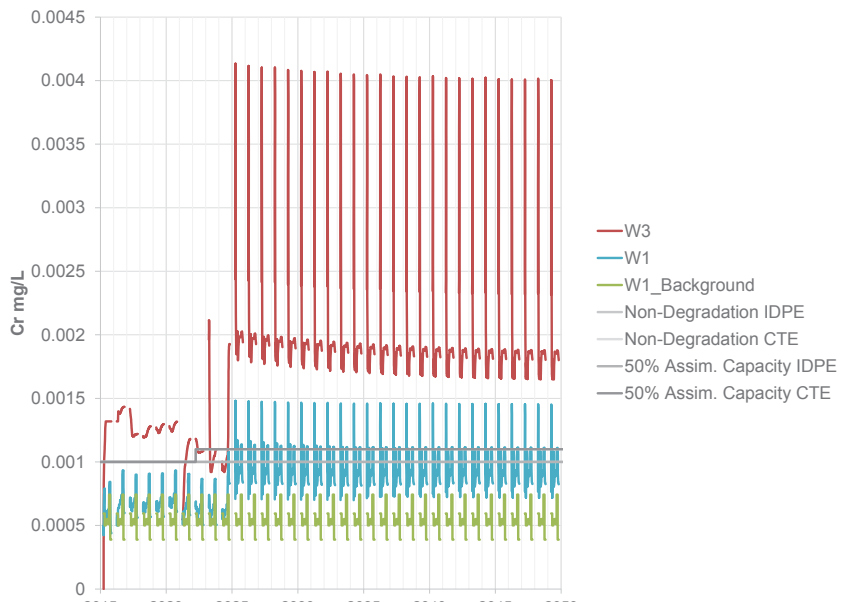


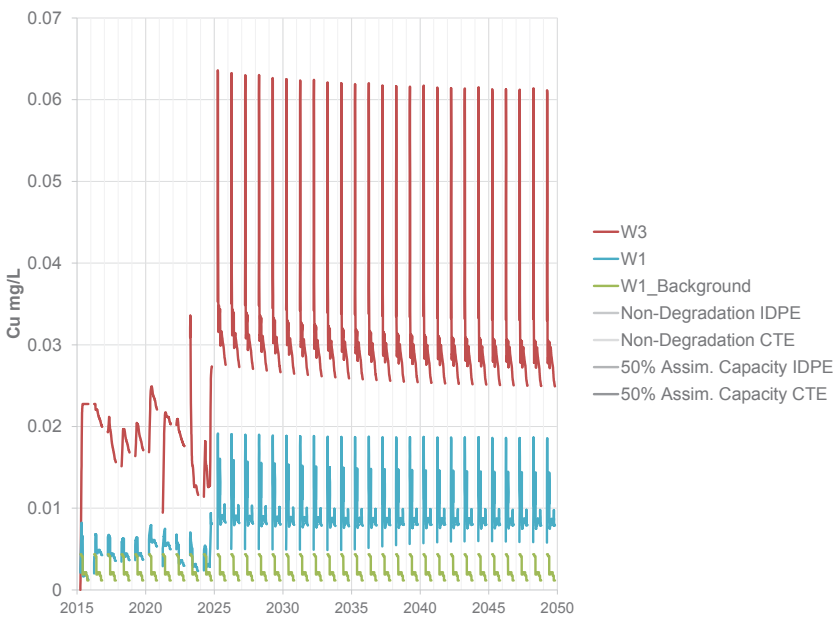


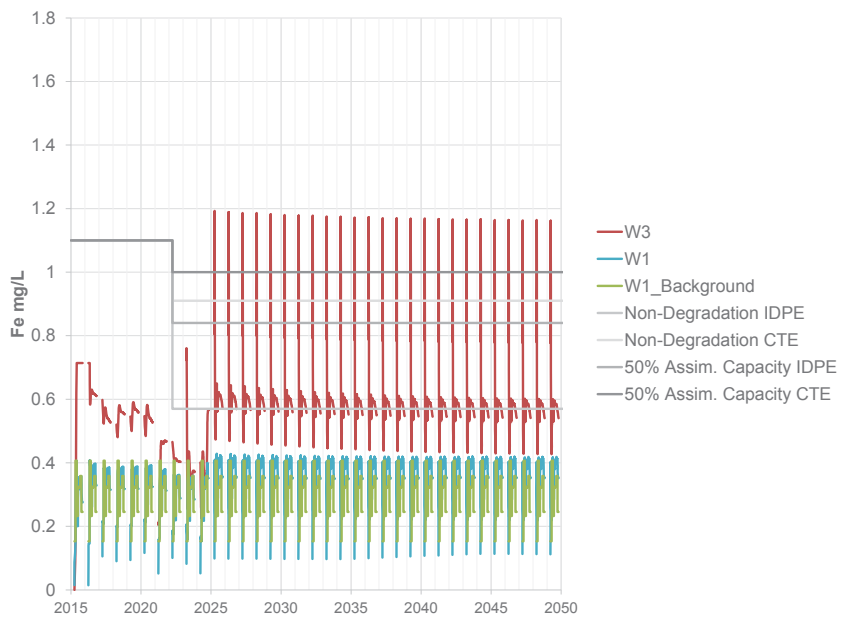




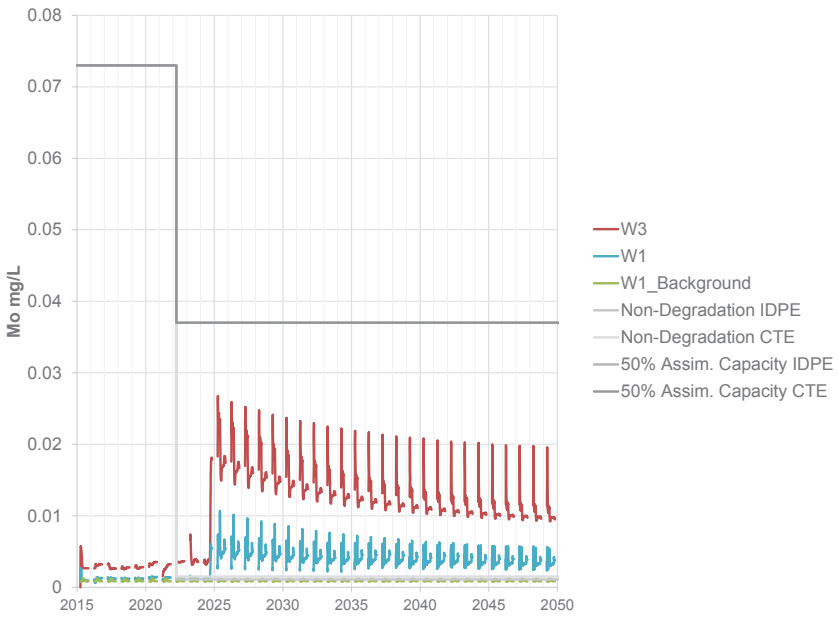


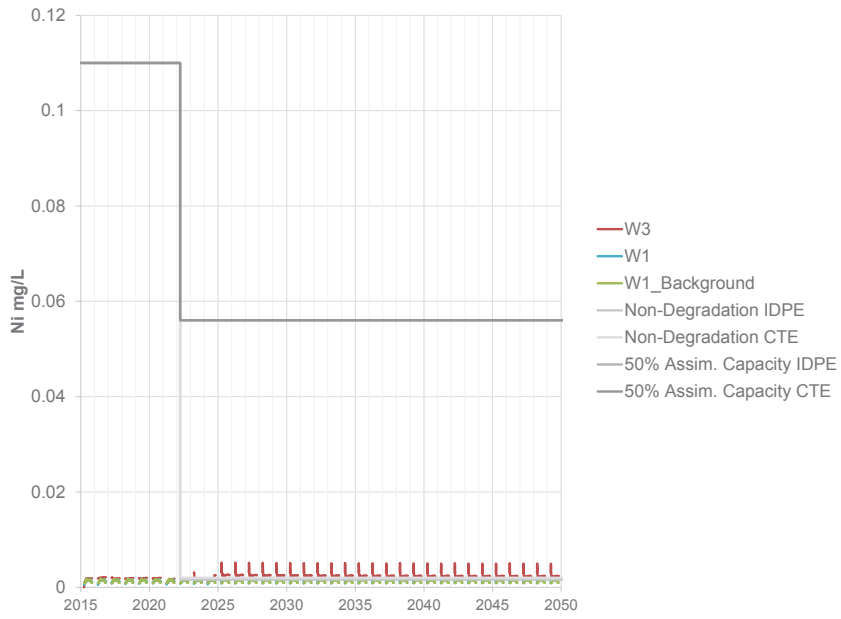


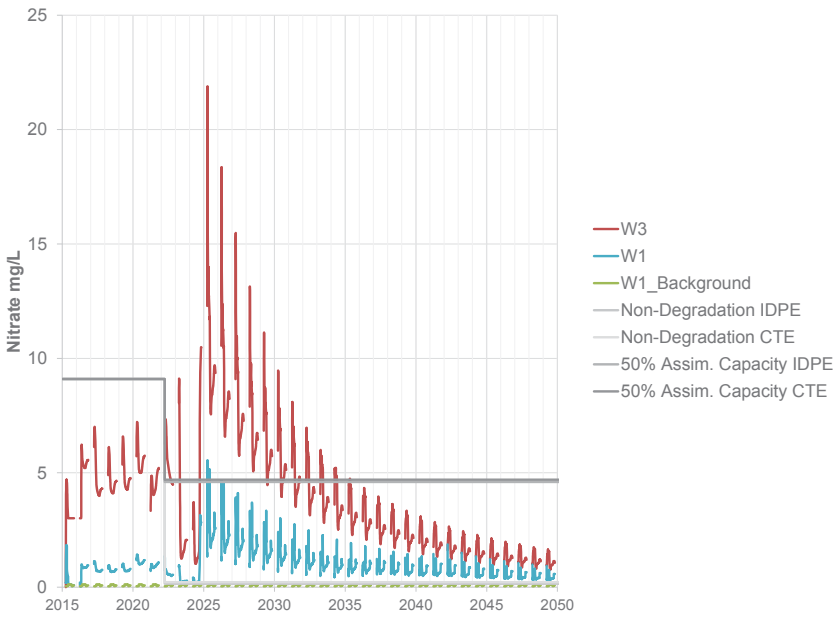


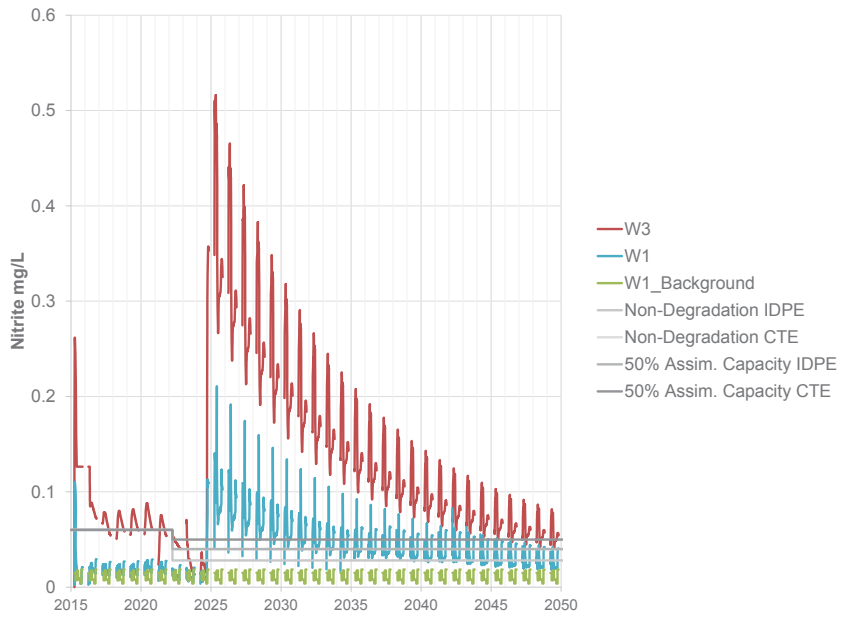


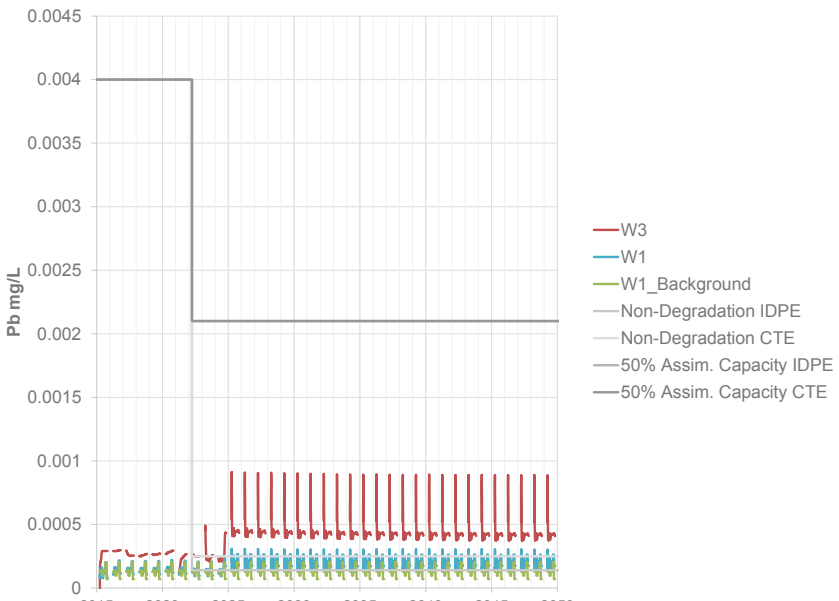
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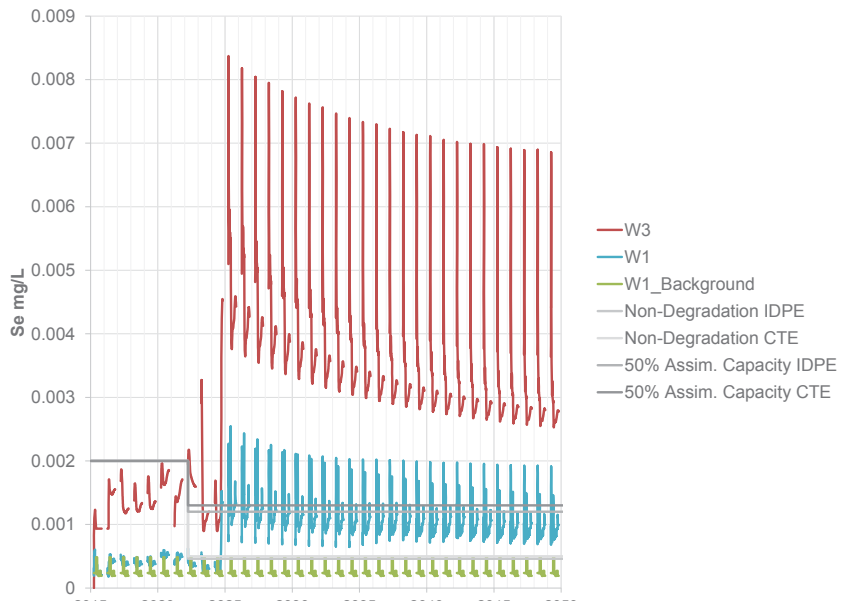


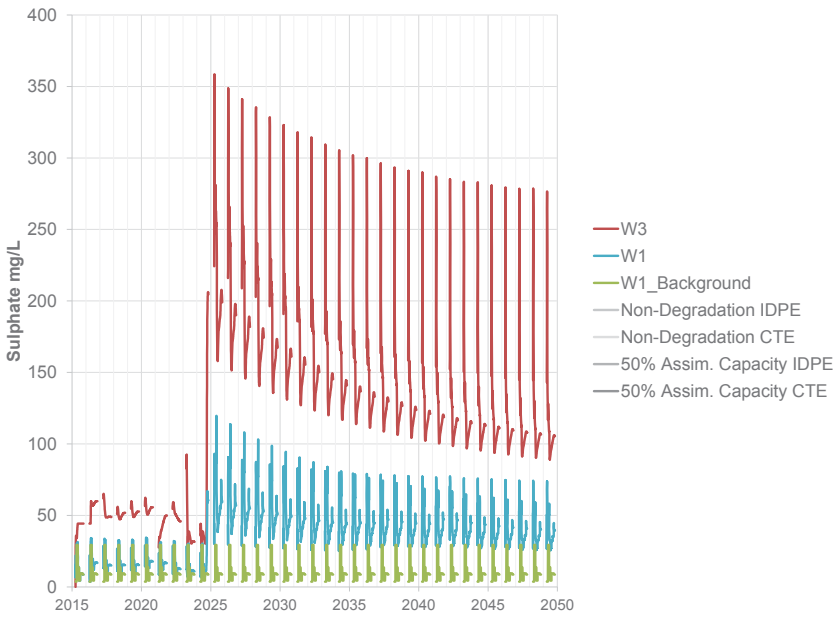


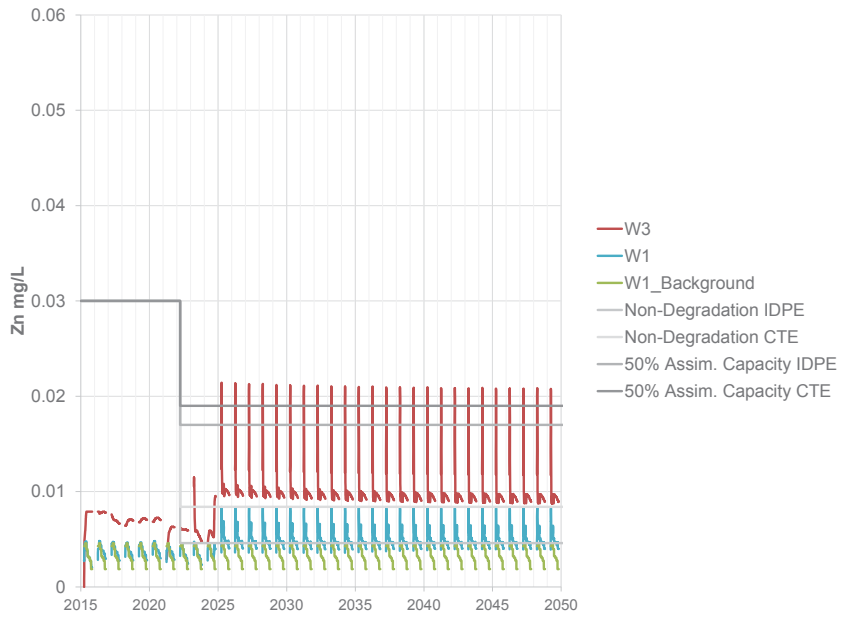




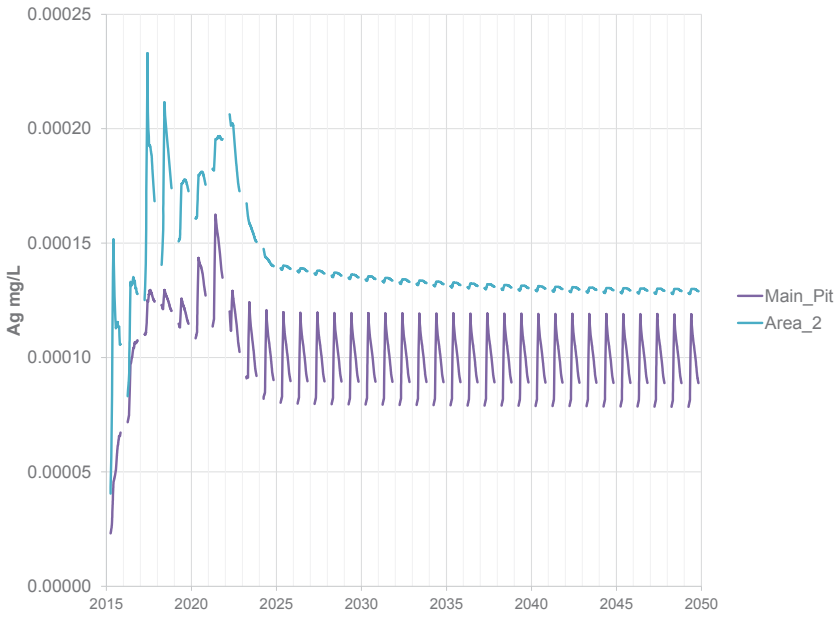


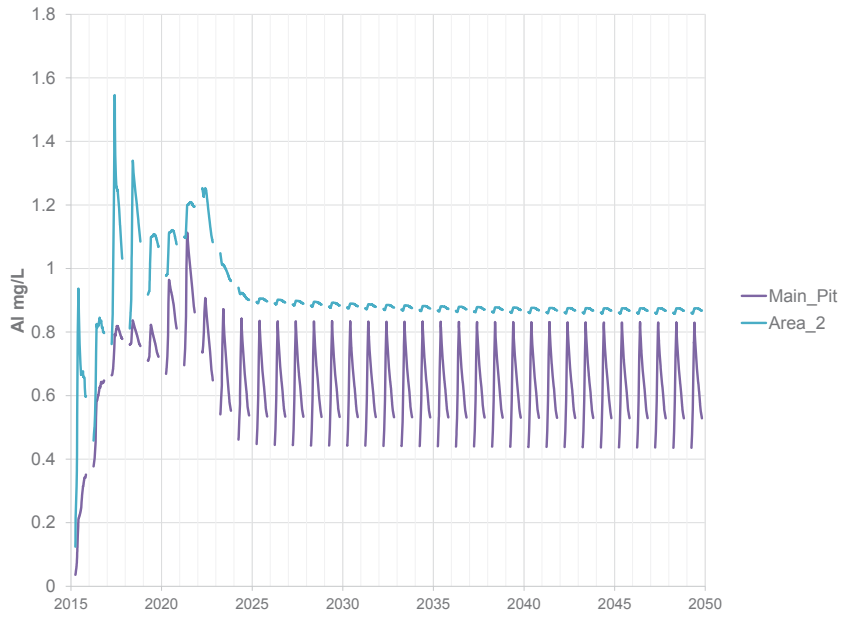


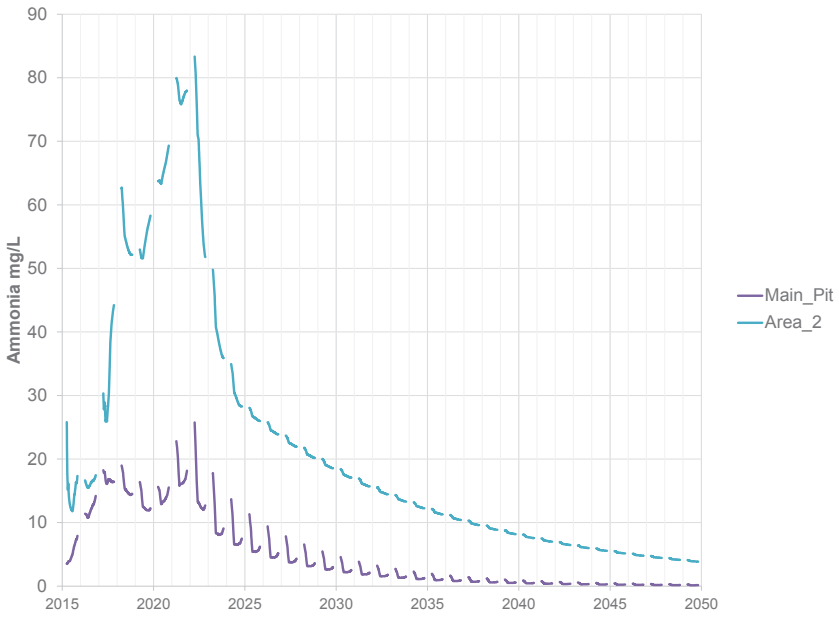


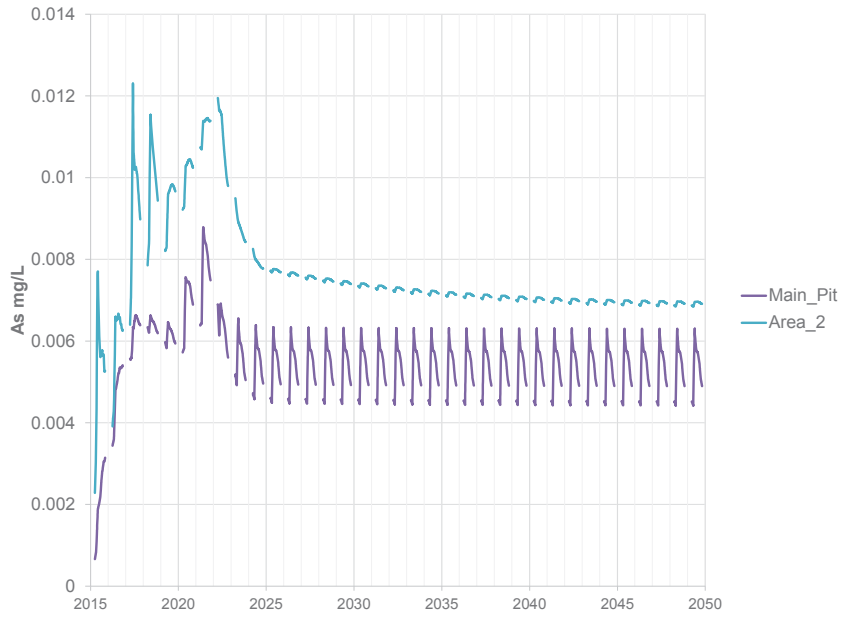


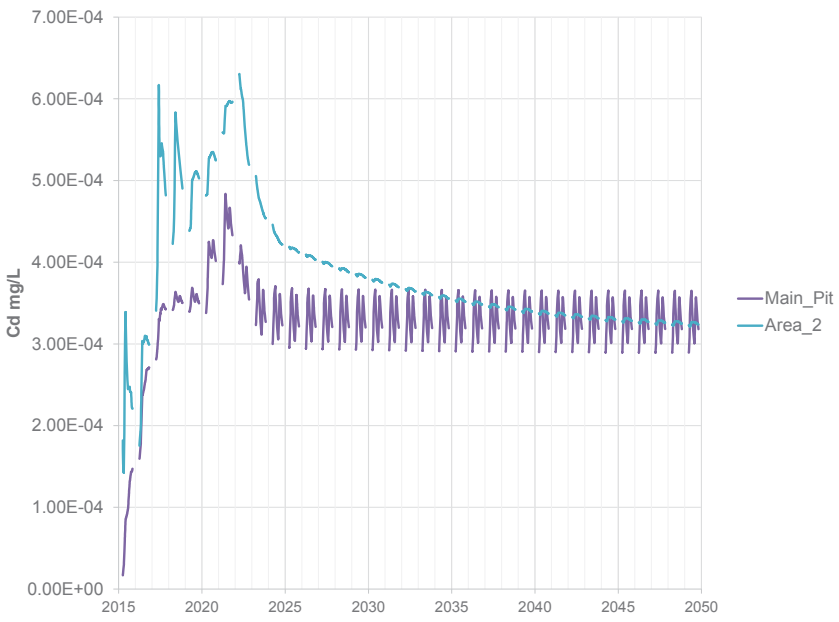
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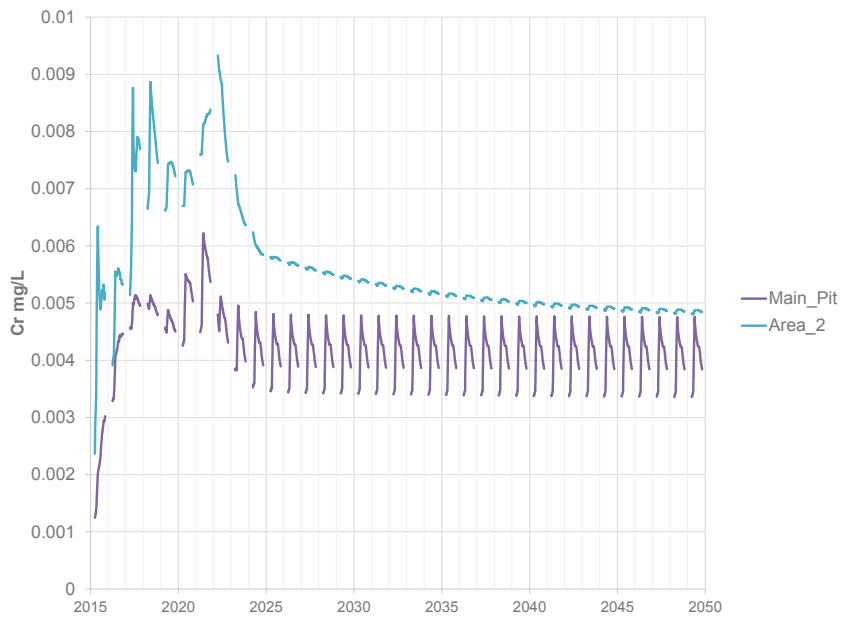

Ag mg/L
0.00025
0.00020
0.00015
0.00010
0.00005
0.00000
2015
2020
2025
2030
2035
2040
2045
2050
Main_Pit
Area_2

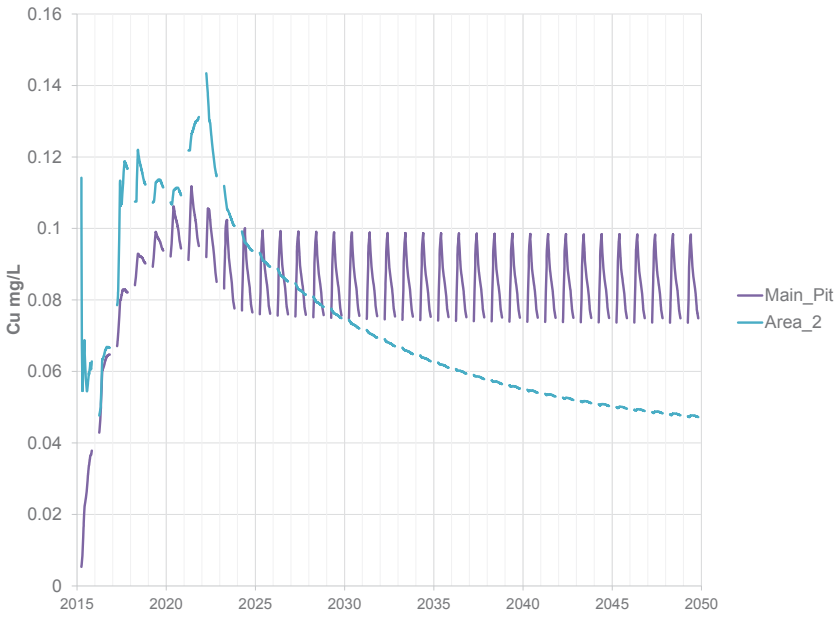

Al mg/L
1.8
1.6
1.4
1.2
1
0.8
0.6
0.4
0.2
0
2015
2020
2025
2030
2035
2040
2045
2050
Main_Pit
Area_2

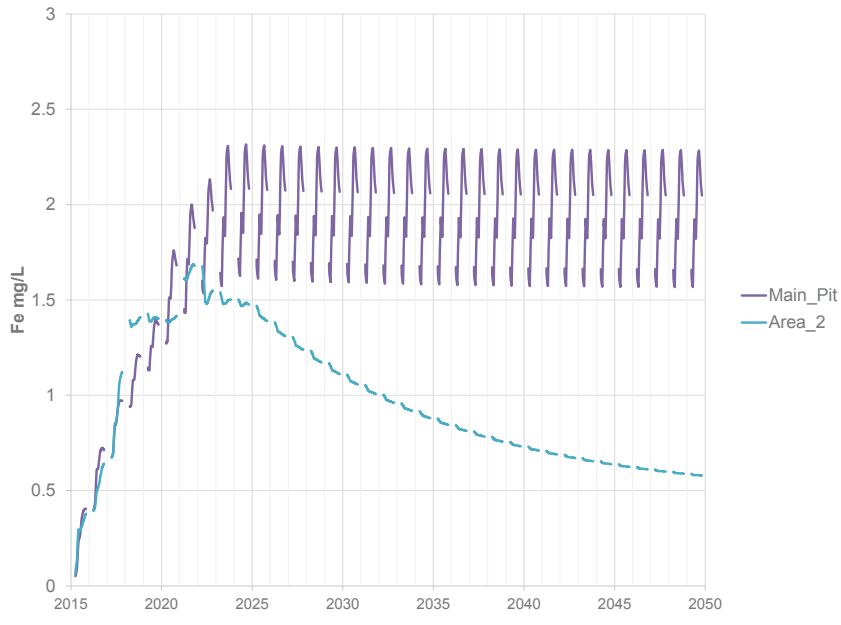

Ammonia mg/L
90
80
70
60
50
40
30
20
10
0
2015
2020
2025
2030
2035
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2045
2050
Main_Pit
Area_2

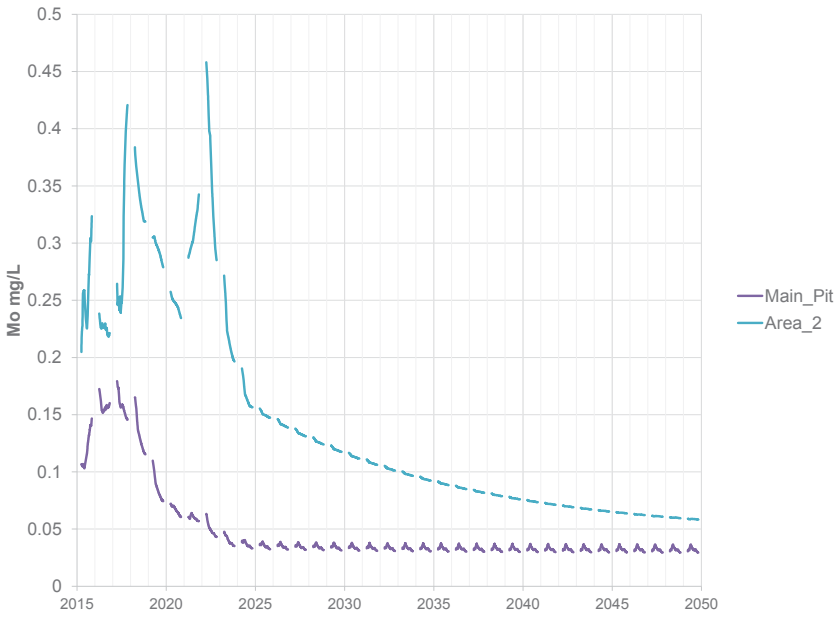

As mg/L
0.014
0.012
0.01
0.008
0.006
0.004
0.002
0
2015
2020
2025
2030
2035
2040
2045
2050
Main_Pit
Area_2

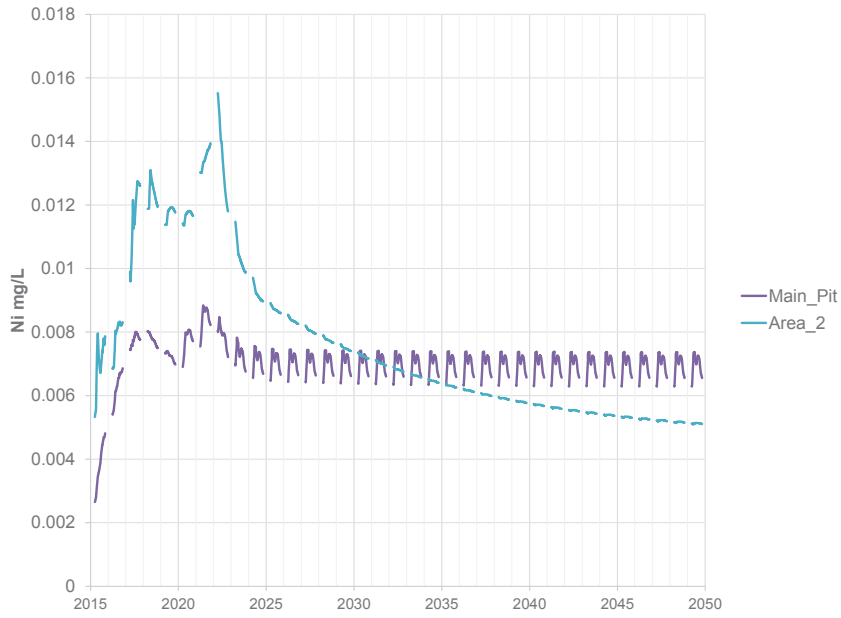

Cd mg/L
7.00E-04
6.00E-04
5.00E-04
4.00E-04
3.00E-04
2.00E-04
1.00E-04
0.00E+00
2015
2020
2025
2030
2035
2040
2045
2050
Main_Pit
Area_2

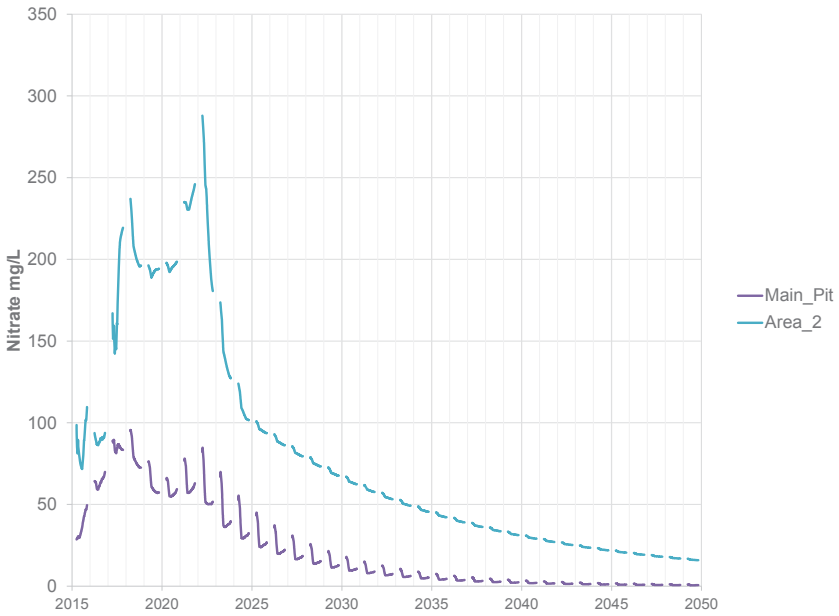

Cr mg/L
0.01
0.009
0.008
0.007
0.006
0.005
0.004
0.003
0.002
0.001
0
2015
2020
2025
2030
2035
2040
2045
2050
Main_Pit
Area_2

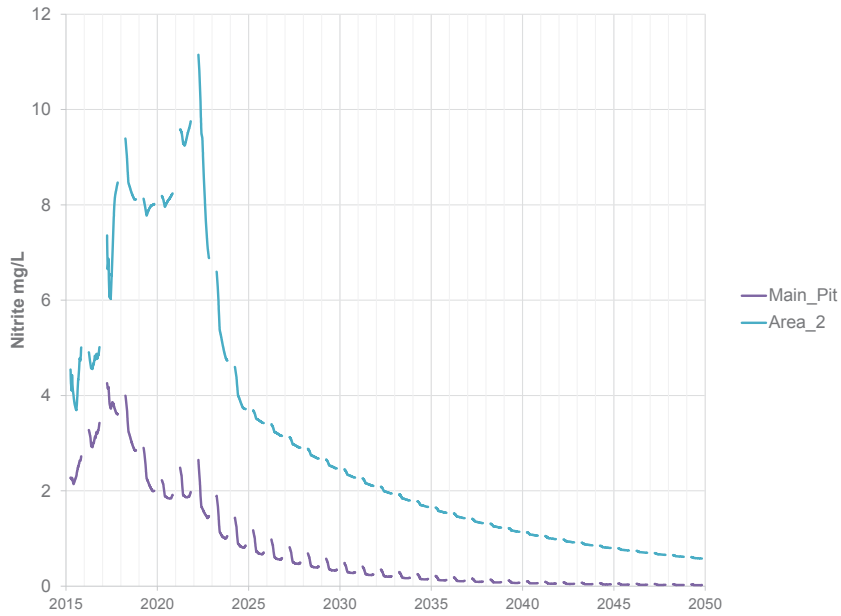

Cu mg/L
0.16
0.14
0.12
0.1
0.08
0.06
0.04
0.02
0
2015
2020
2025
2030
2035
2040
2045
2050
Main_Pit
Area_2

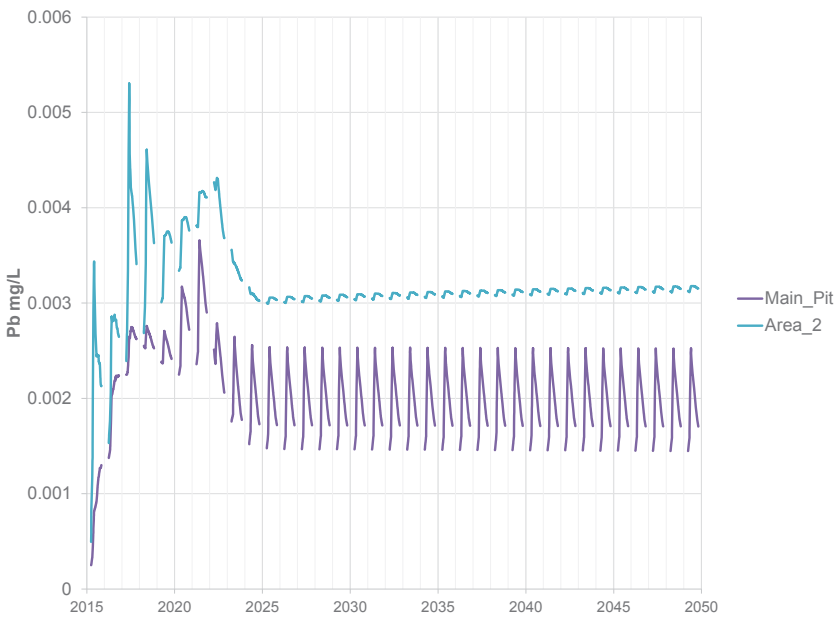

Fe mg/L
3
2.5
2
1.5
1
0.5
0
2015
2020
2025
2030
2035
2040
2045
2050
Main_Pit
Area_2

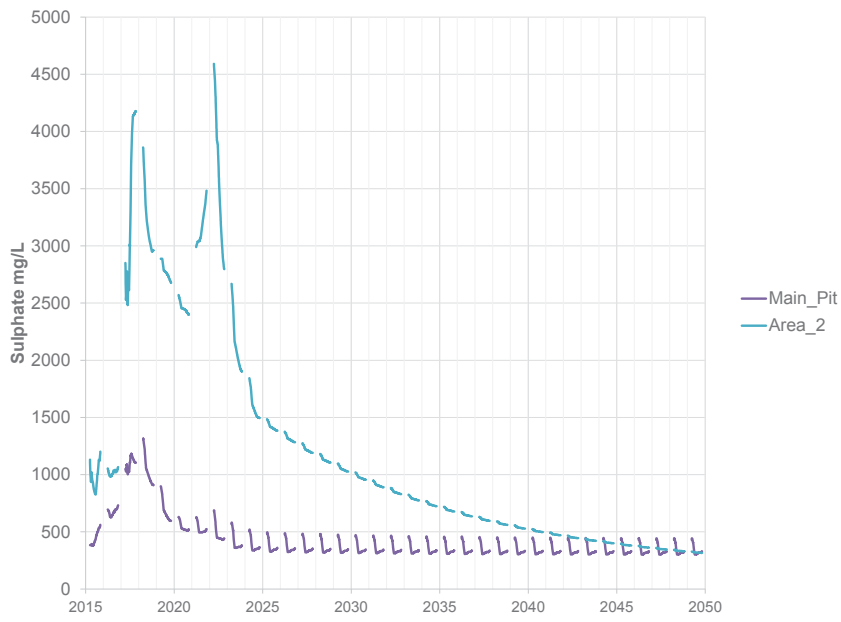


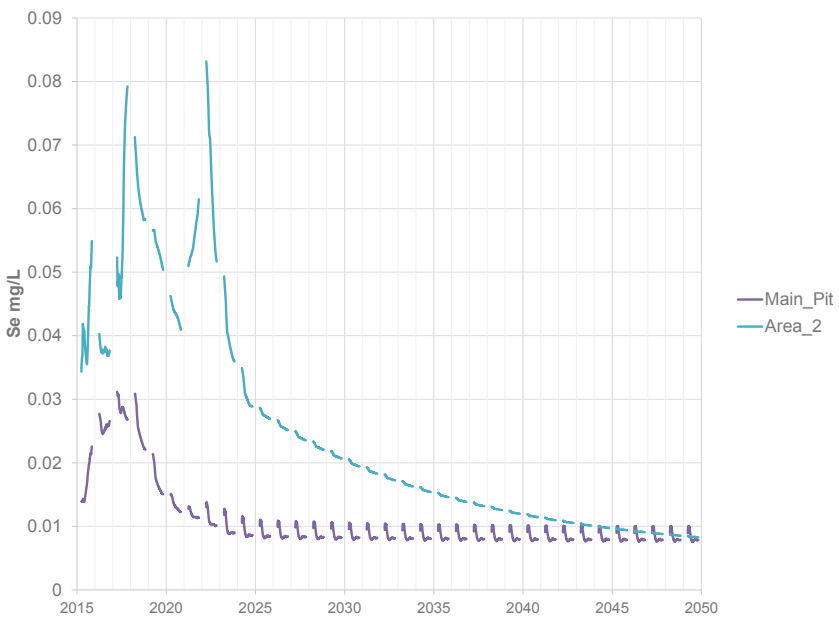


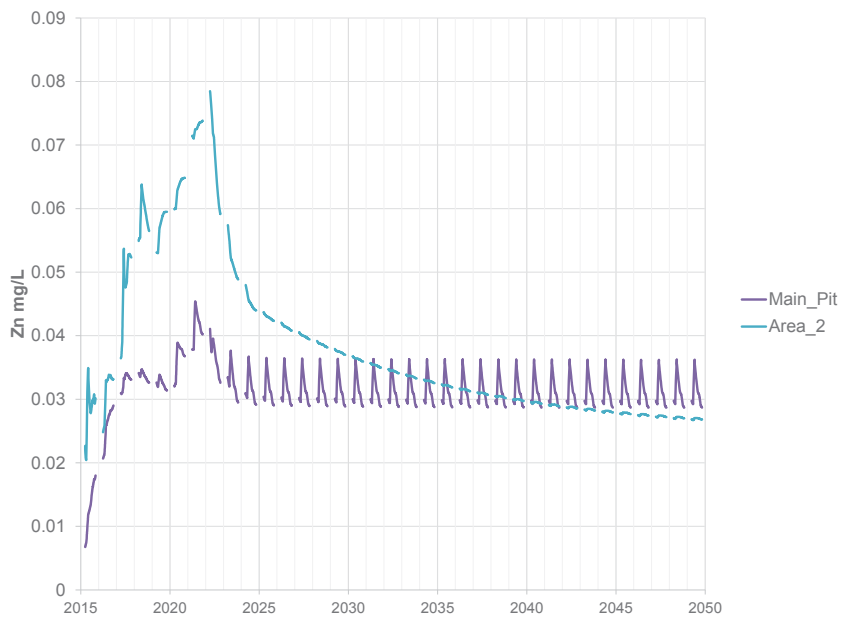




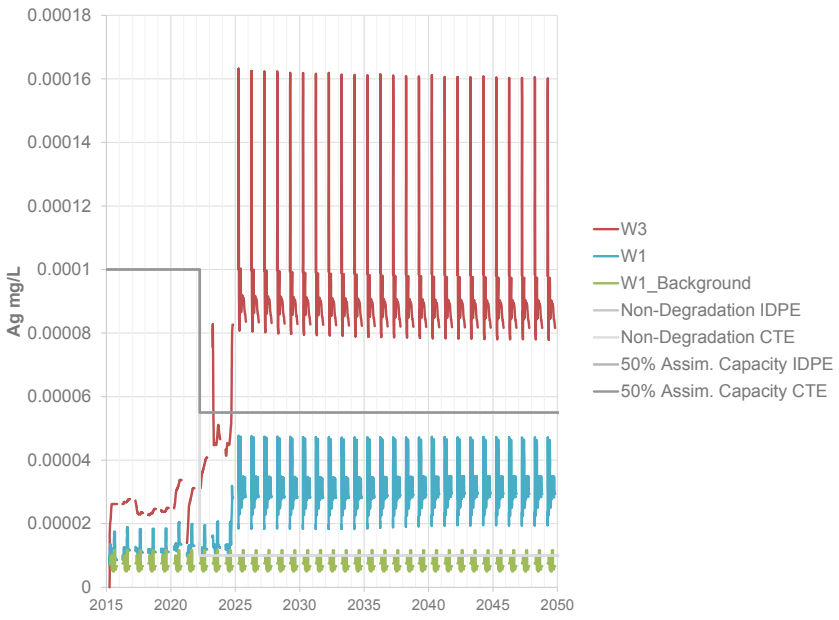


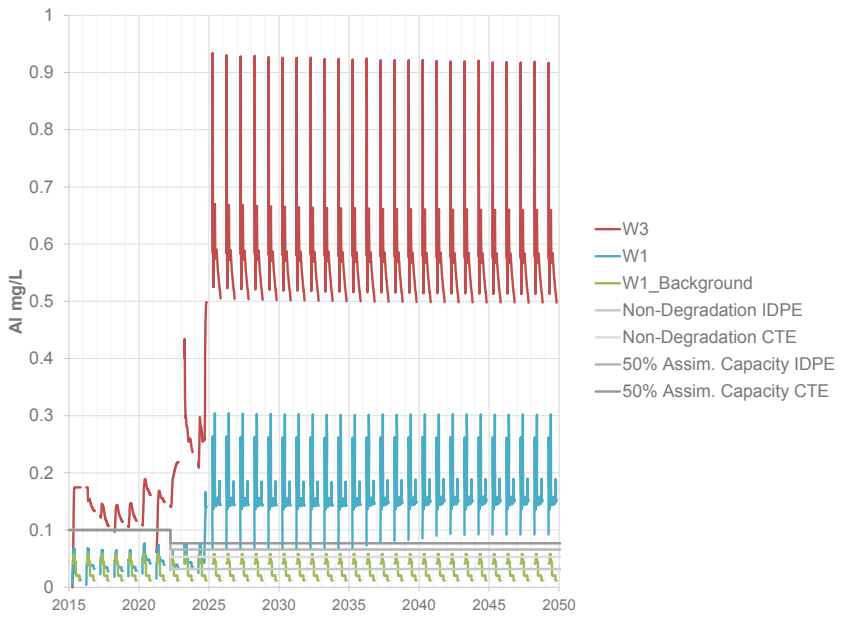


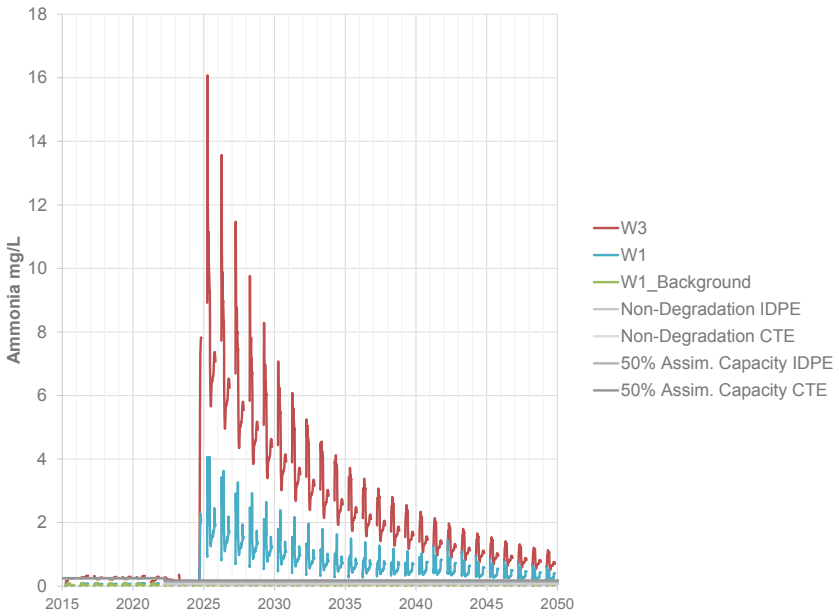


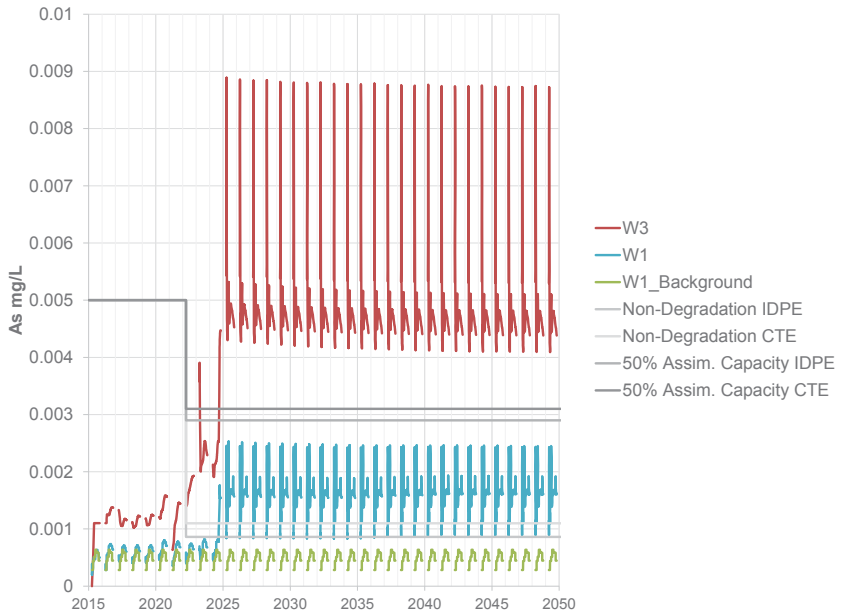


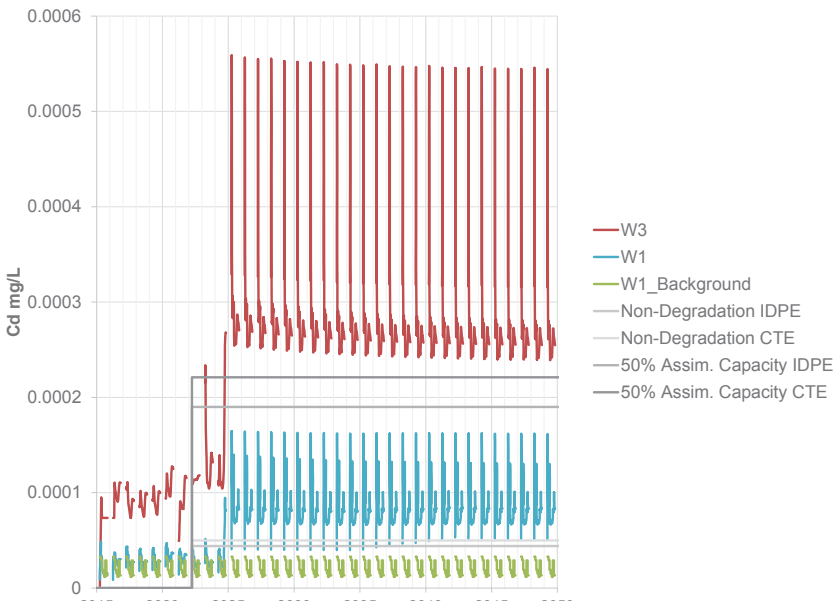
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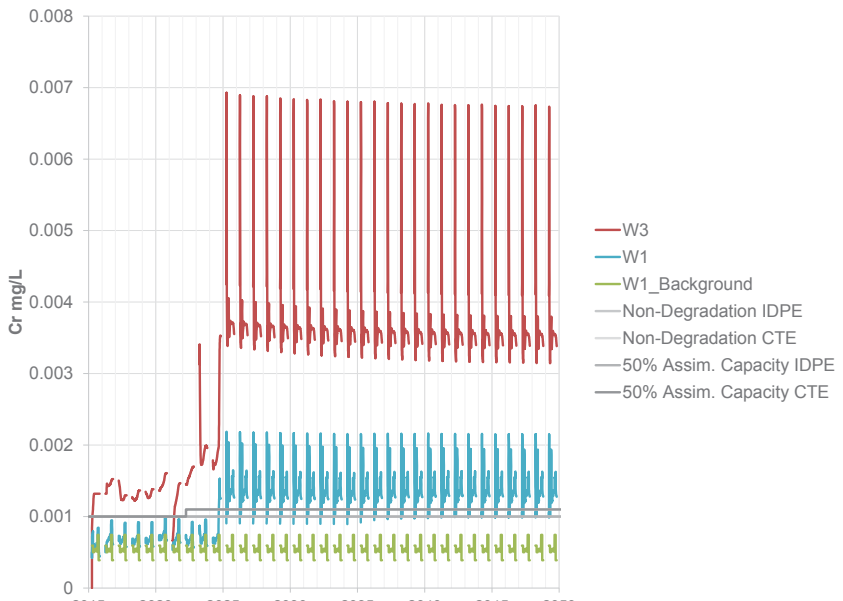


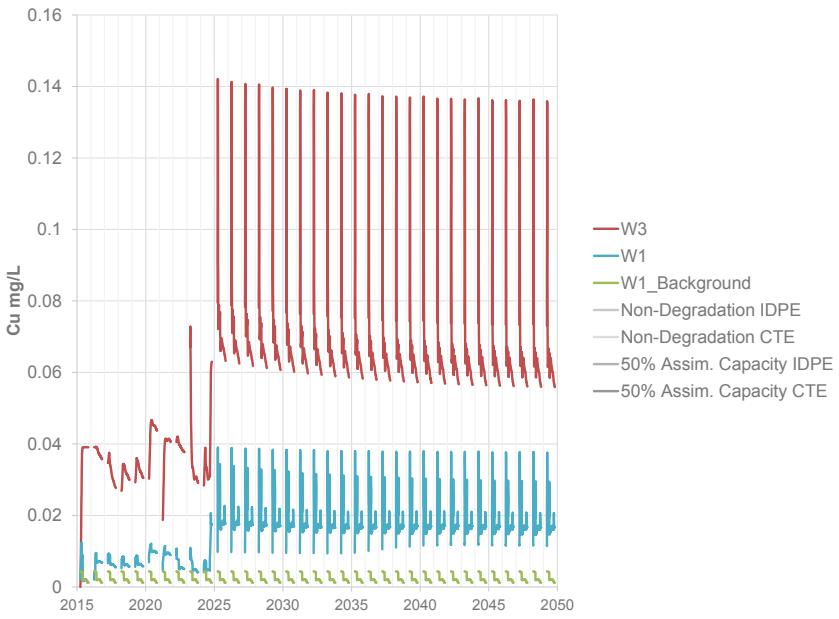


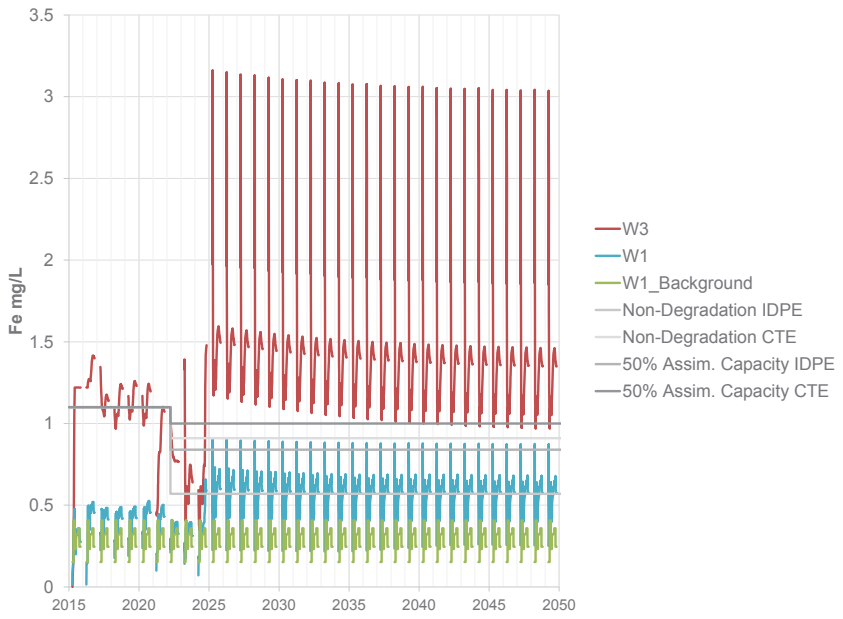




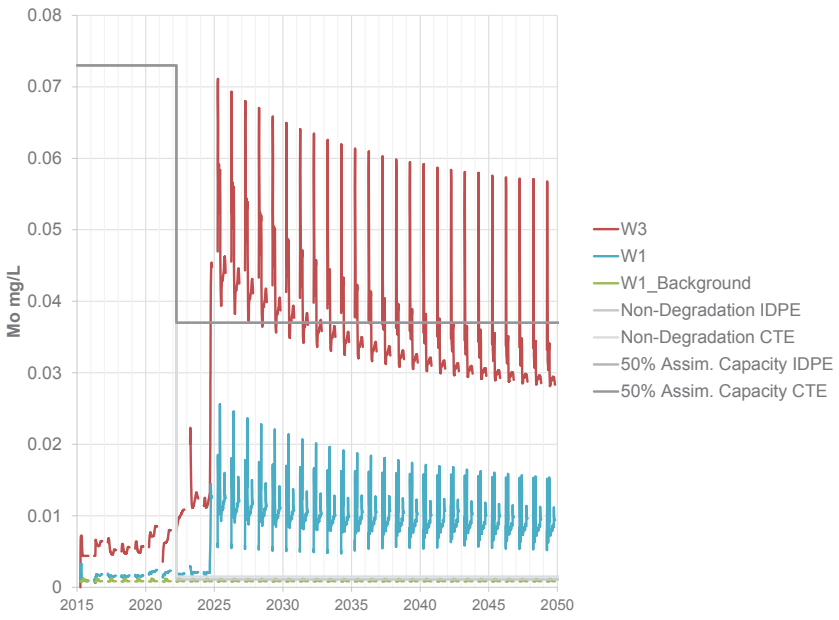


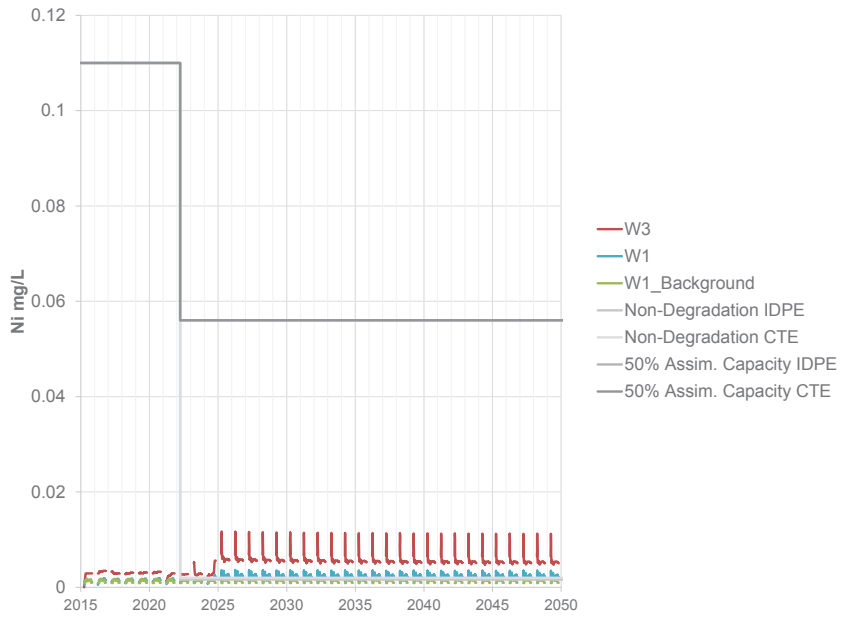


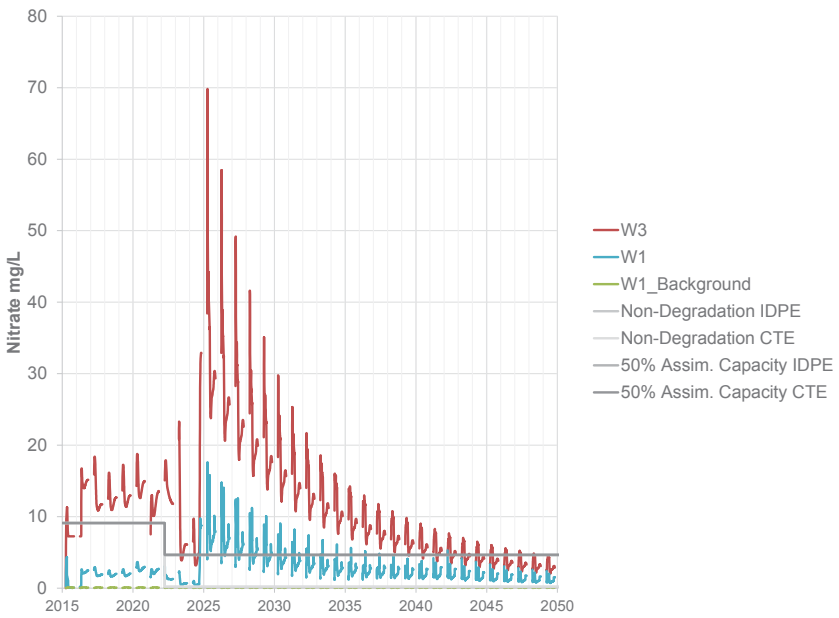


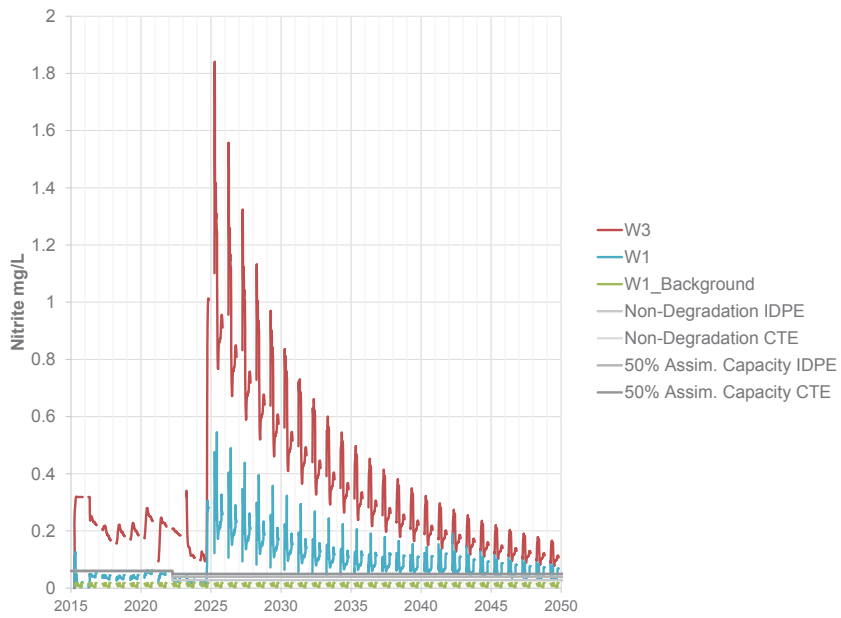


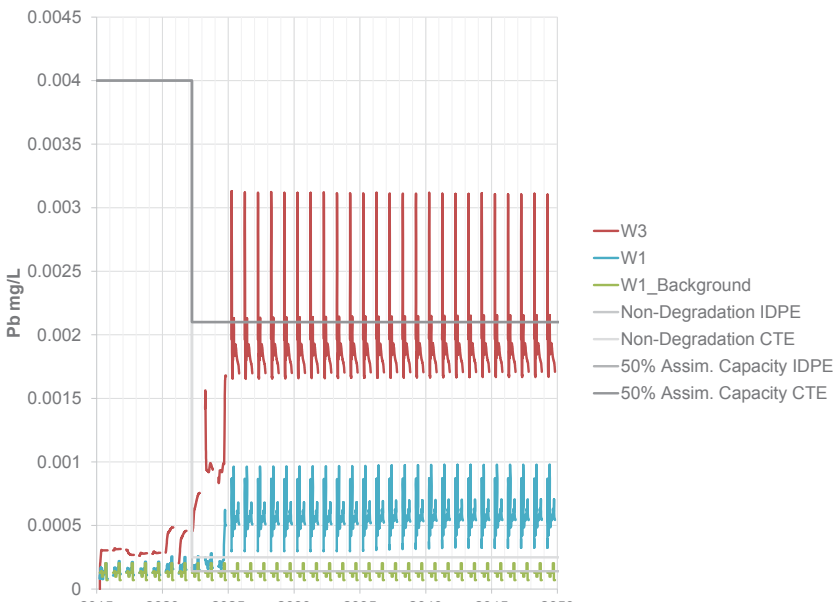
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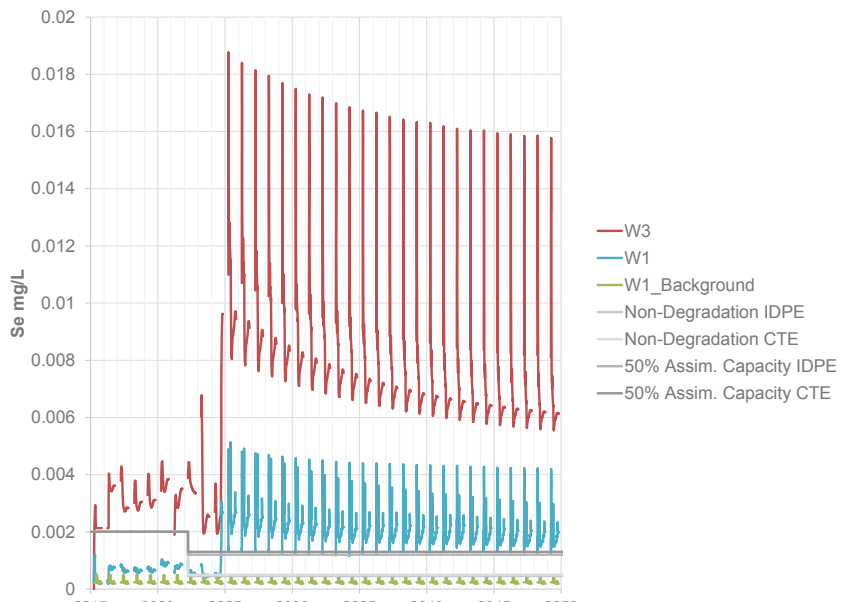

Mo mg/L
0.08
0.07
0.06
0.05
0.04
0.03
0.02
0.01
0
2015
2020
2025
2030
2035
2040
2045
2050
W3
W1
W1_Background
Non-Degradation IDPE
Non-Degradation CTE
50% Assim. Capacity IDPE
50% Assim. Capacity CTE

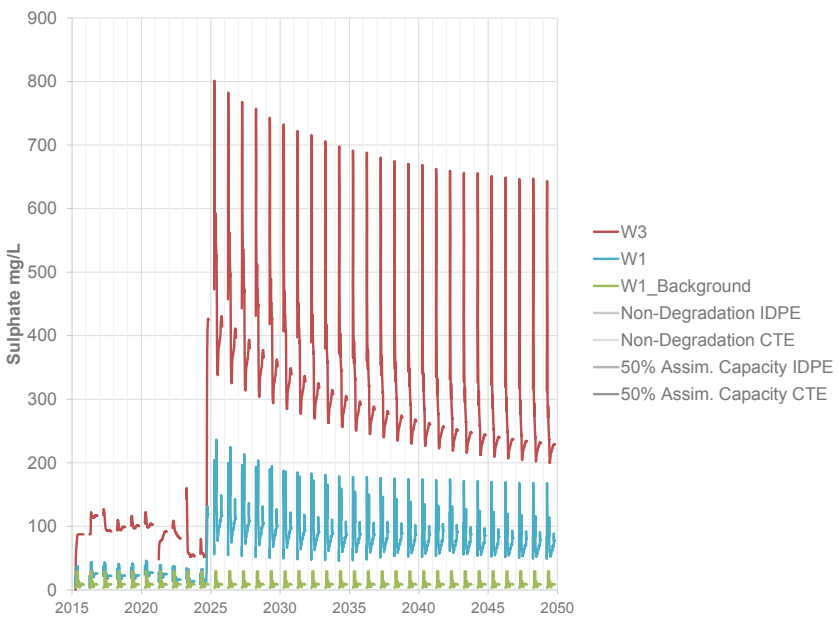

Ni mg/L
0.12
0.1
0.08
0.06
0.04
0.02
0
2015
2020
2025
2030
2035
2040
2045
2050
W3
W1
W1_Background
Non-Degradation IDPE
Non-Degradation CTE
50% Assim. Capacity IDPE
50% Assim. Capacity CTE

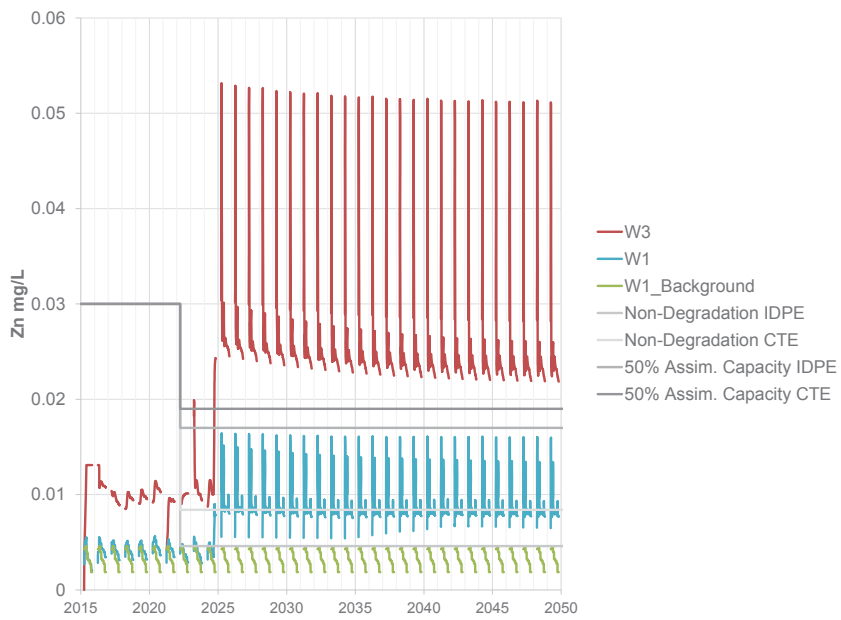

Nitrate mg/L
80
70
60
50
40
30
20
10
0
2015
2020
2025
2030
2035
2040
2045
2050
W3
W1
W1_Background
Non-Degradation IDPE
Non-Degradation CTE
50% Assim. Capacity IDPE
50% Assim. Capacity CTE

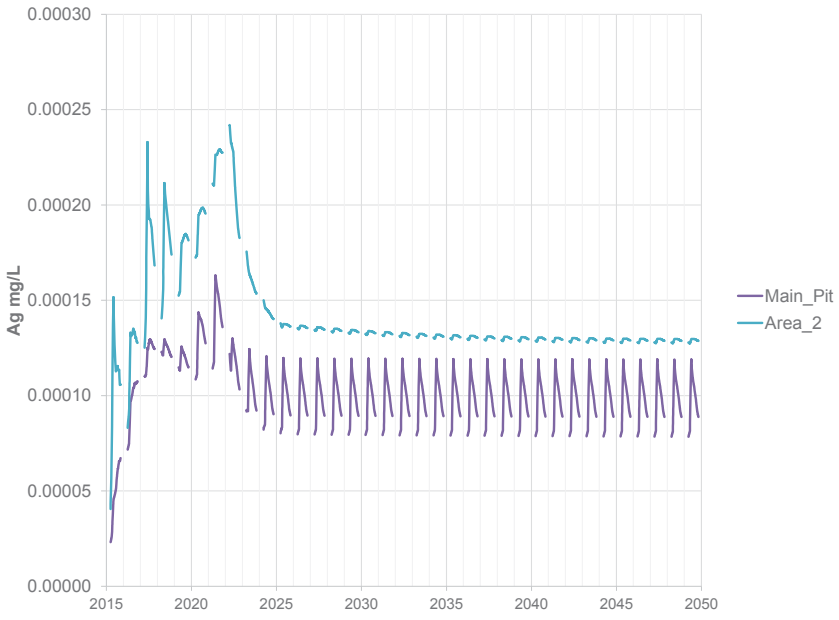

Nitrite mg/L
2
1.8
1.6
1.4
1.2
1
0.8
0.6
0.4
0.2
0
2015
2020
2025
2030
2035
2040
2045
2050
W3
W1
W1_Background
Non-Degradation IDPE
Non-Degradation CTE
50% Assim. Capacity IDPE
50% Assim. Capacity CTE

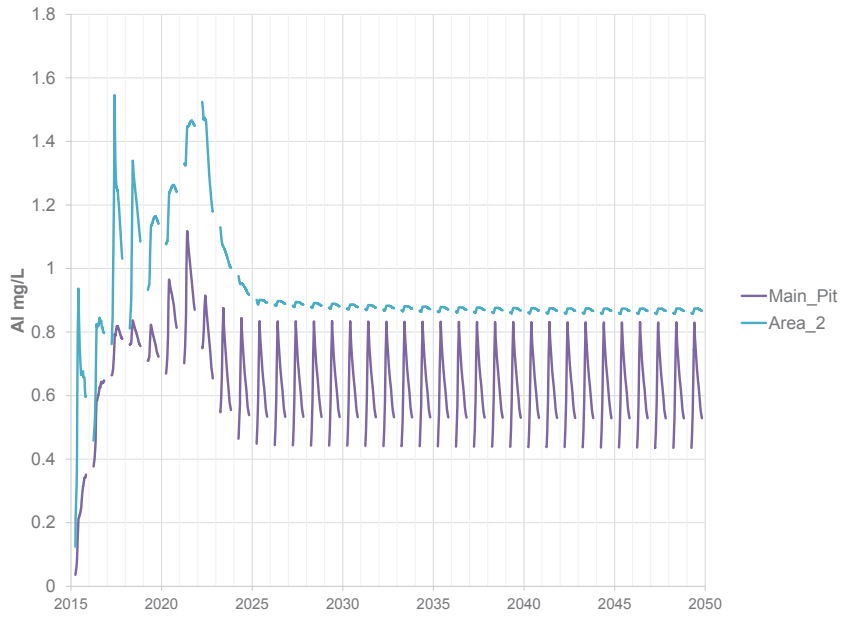

Pb mg/L
0.0045
0.004
0.0035
0.003
0.0025
0.002
0.0015
0.001
0.0005
0
2015
2020
2025
2030
2035
2040
2045
2050
W3
W1
W1_Background
Non-Degradation IDPE
Non-Degradation CTE
50% Assim. Capacity IDPE
50% Assim. Capacity CTE

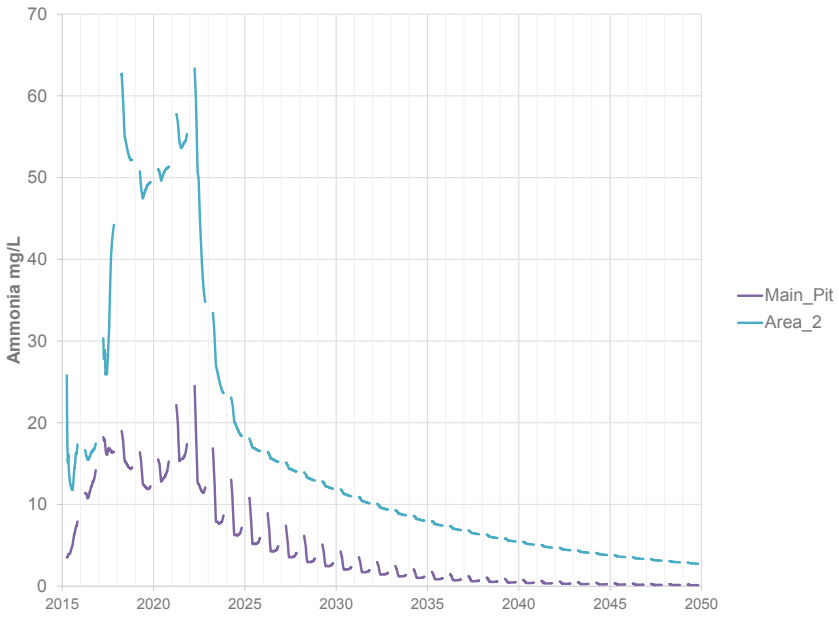

Se mg/L
0.02
0.018
0.016
0.014
0.012
0.01
0.008
0.006
0.004
0.002
0
2015
2020
2025
2030
2035
2040
2045
2050
W3
W1
W1_Background
Non-Degradation IDPE
Non-Degradation CTE
50% Assim. Capacity IDPE
50% Assim. Capacity CTE

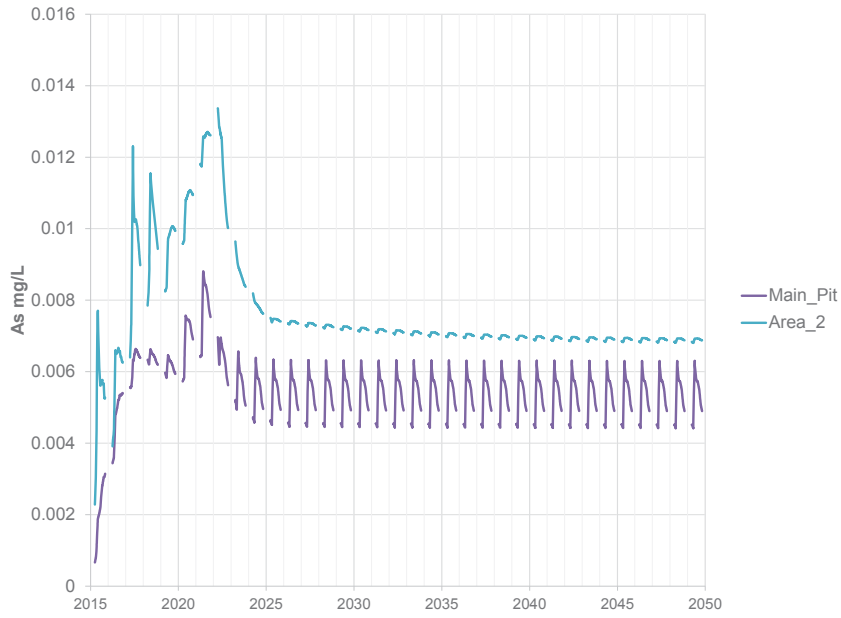

Sulphate mg/L
900
800
700
600
500
400
300
200
100
0
2015
2020
2025
2030
2035
2040
2045
2050
W3
W1
W1_Background
Non-Degradation IDPE
Non-Degradation CTE
50% Assim. Capacity IDPE
50% Assim. Capacity CTE

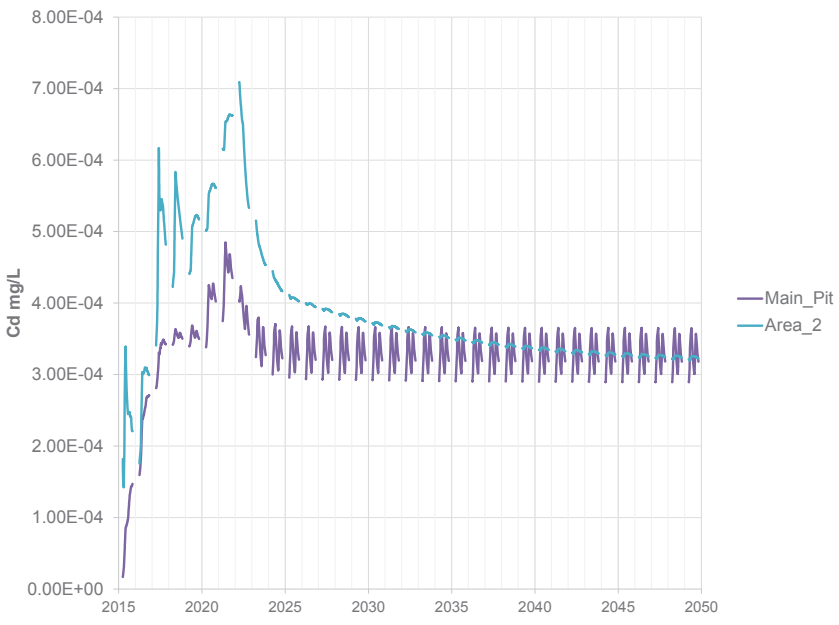

Zn mg/L
0.06
0.05
0.04
0.03
0.02
0.01
0
2015
2020
2025
2030
2035
2040
2045
2050
W3
W1
W1_Background
Non-Degradation IDPE
Non-Degradation CTE
50% Assim. Capacity IDPE
50% Assim. Capacity CTE

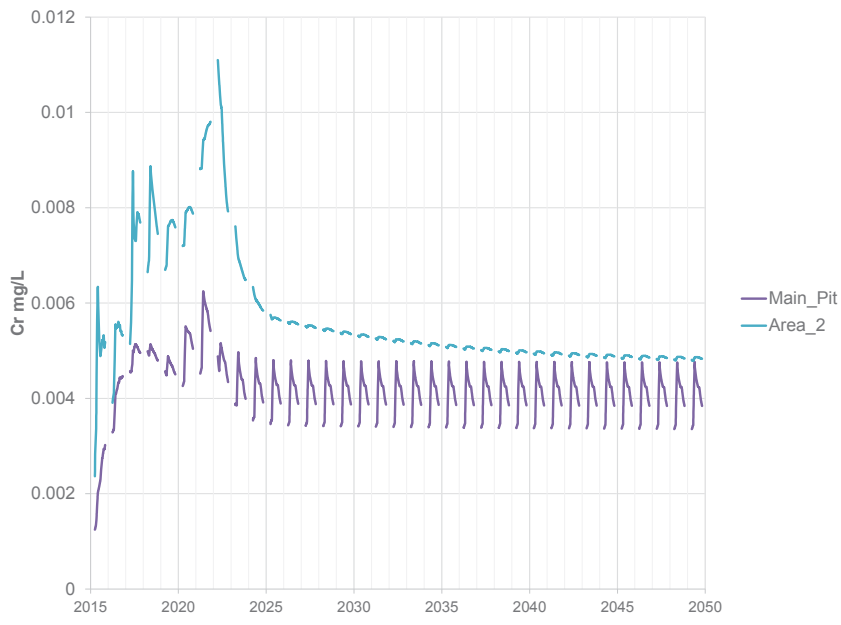


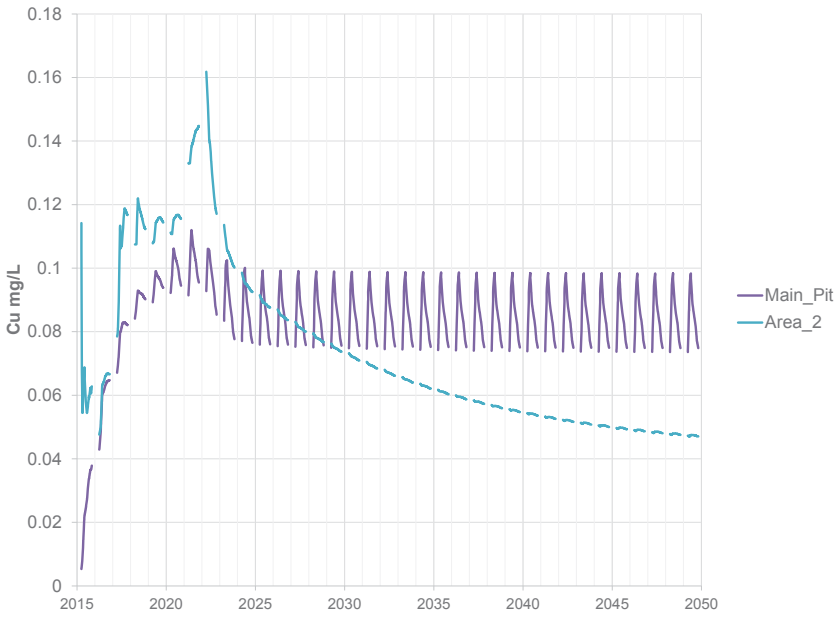


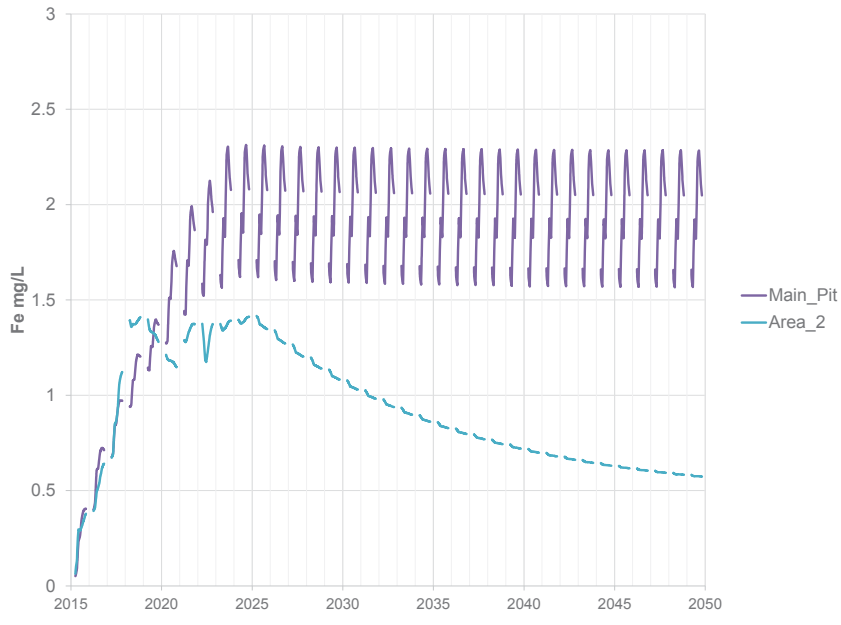




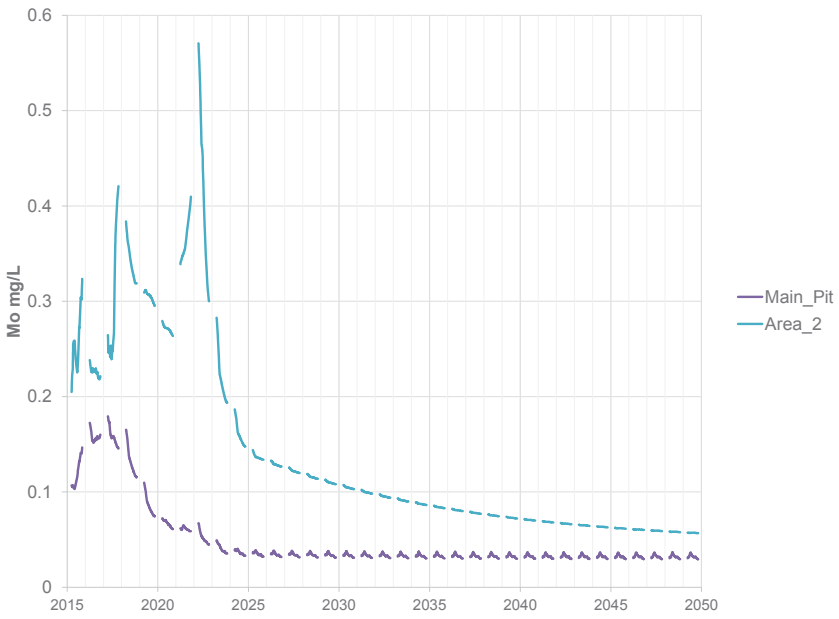


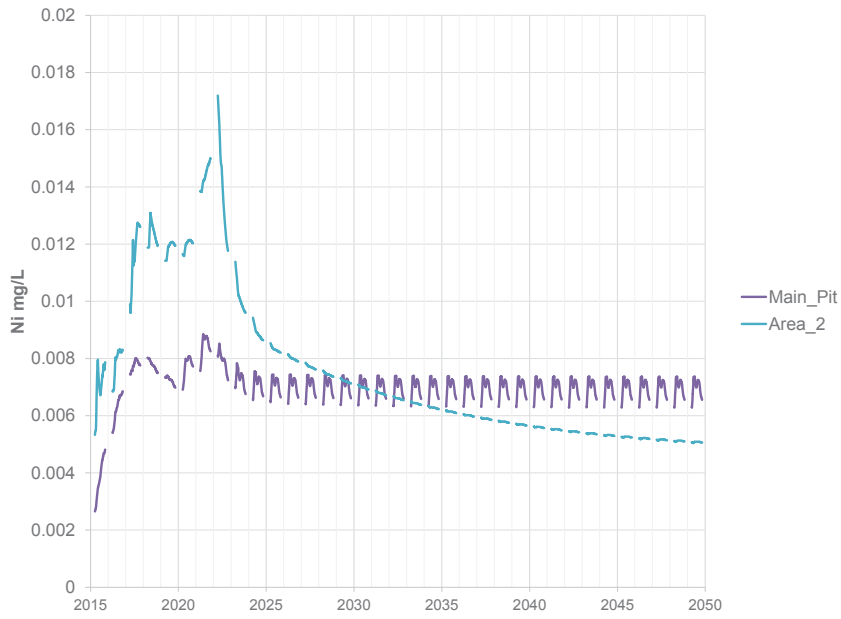


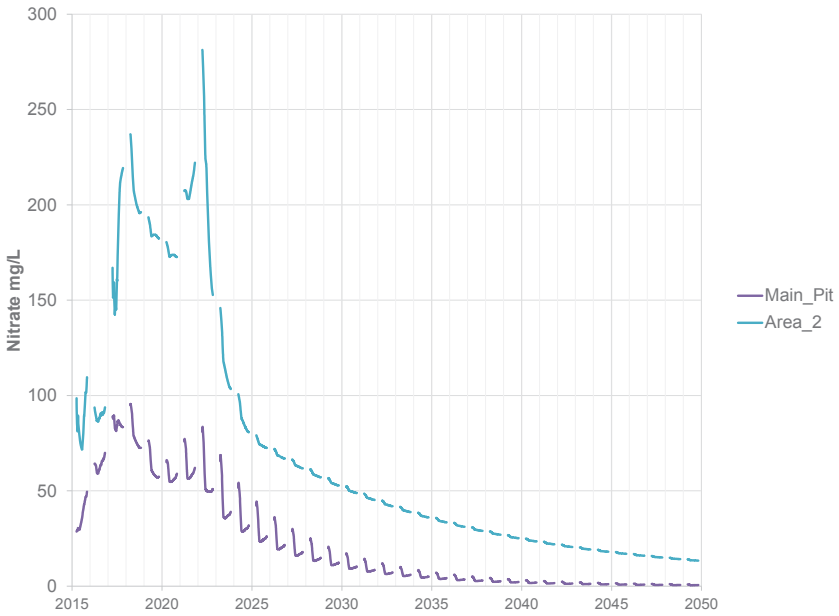


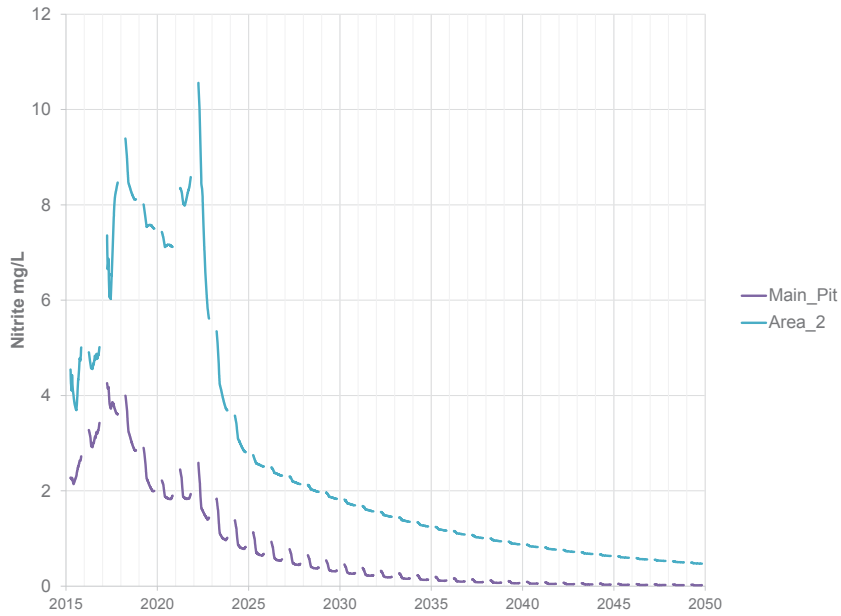


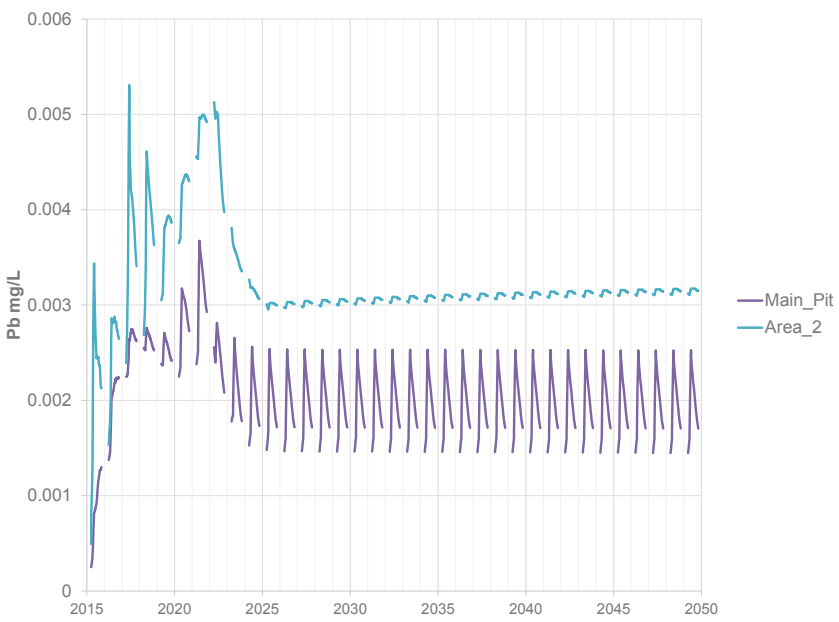
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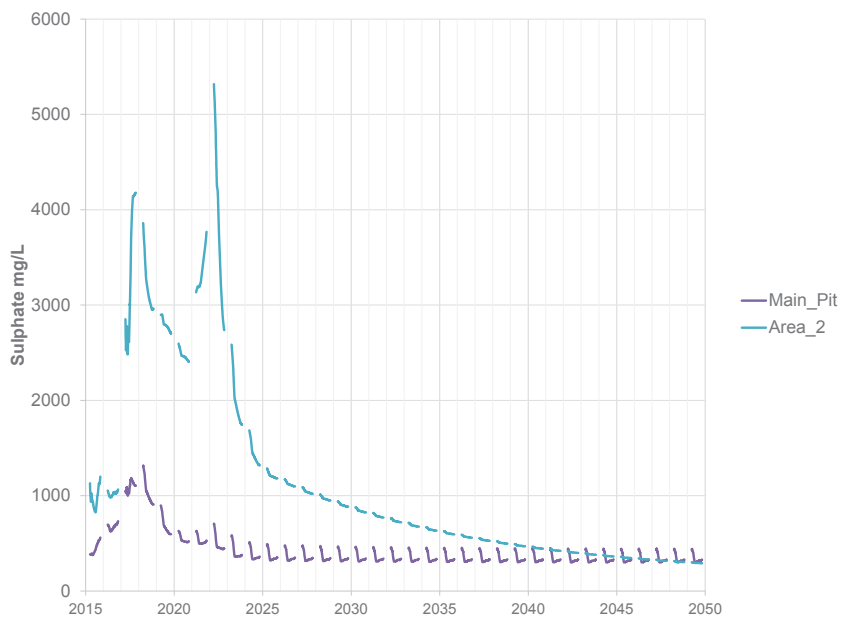


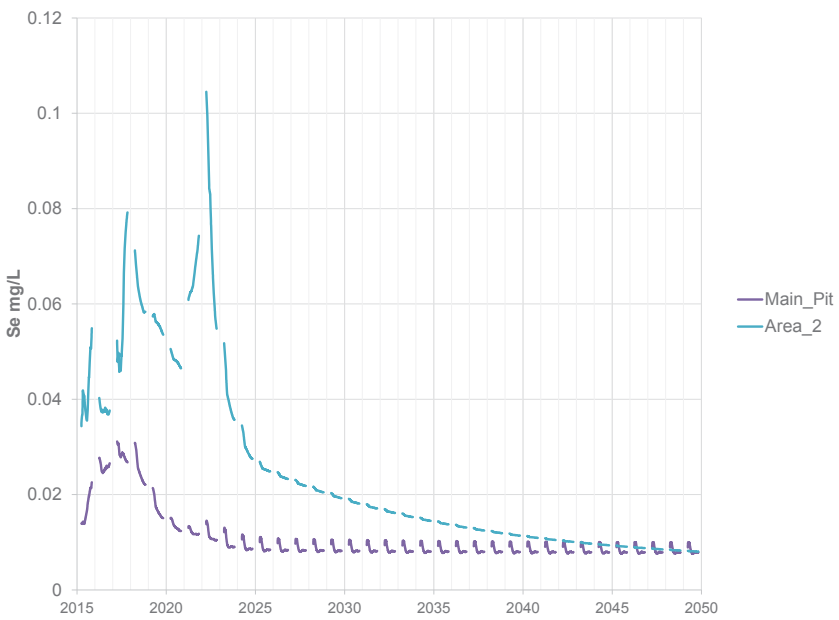


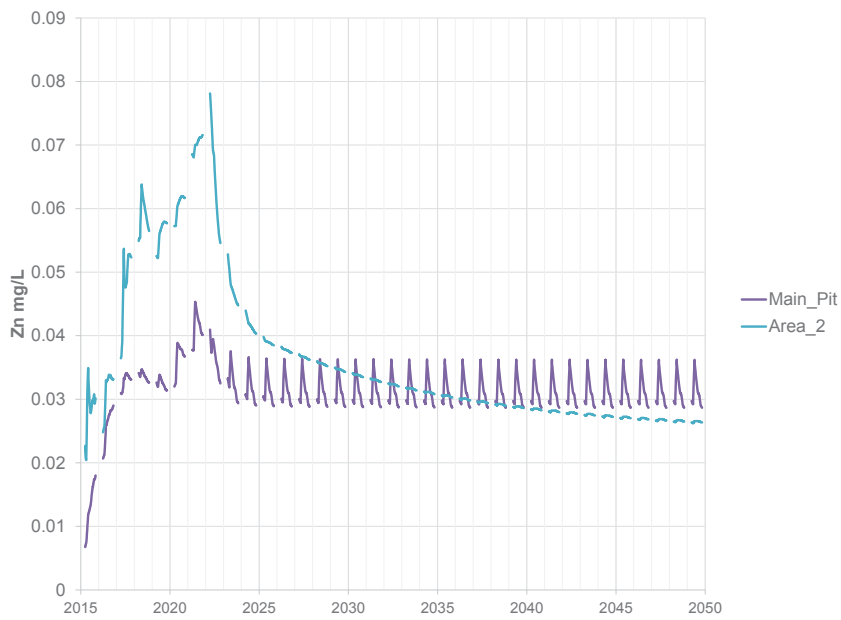




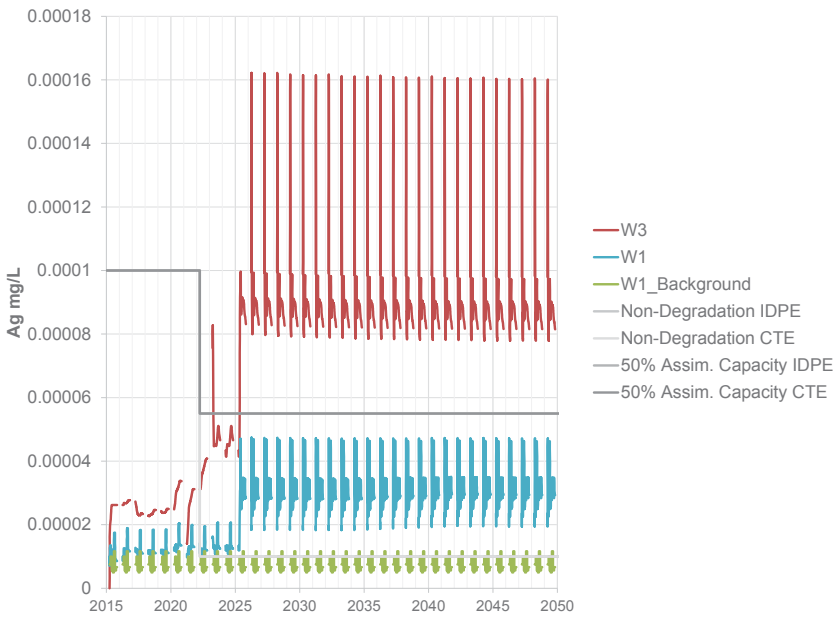


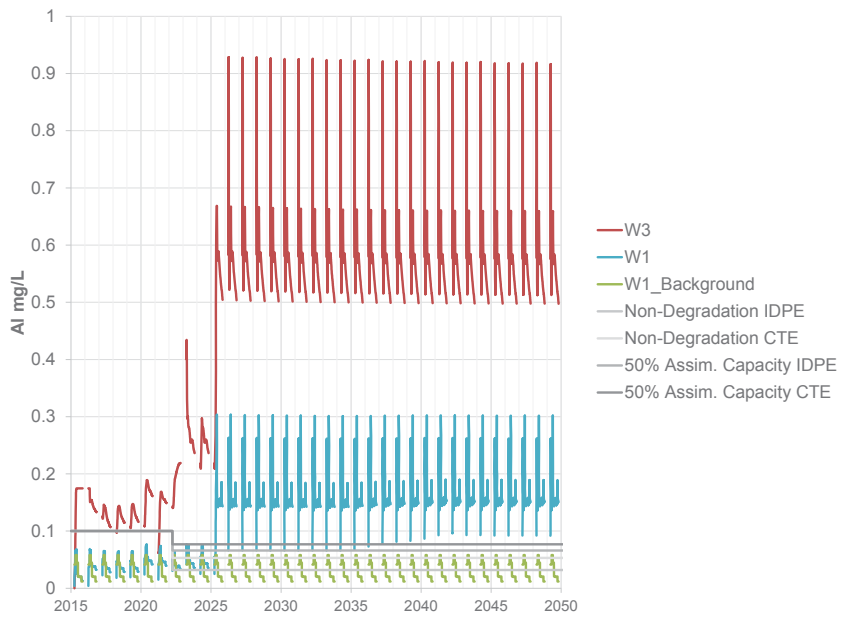


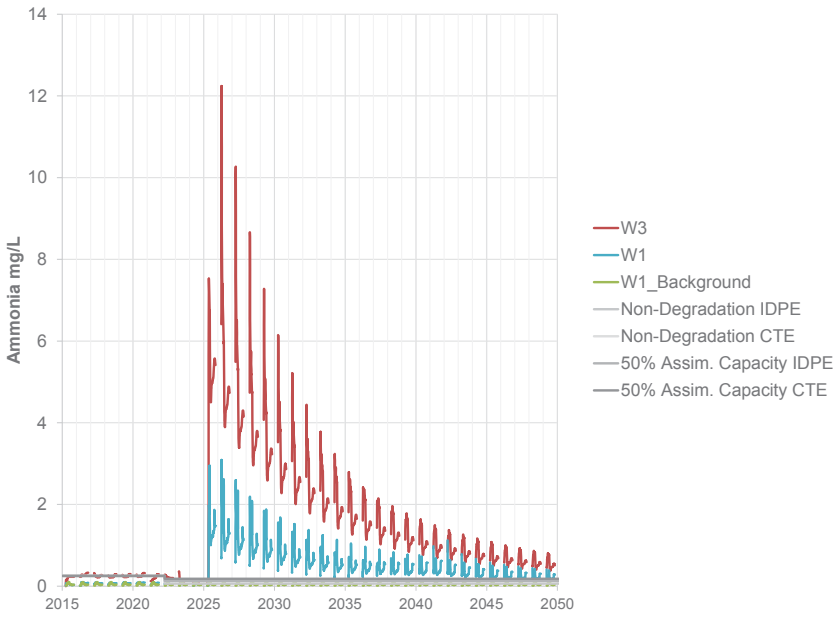


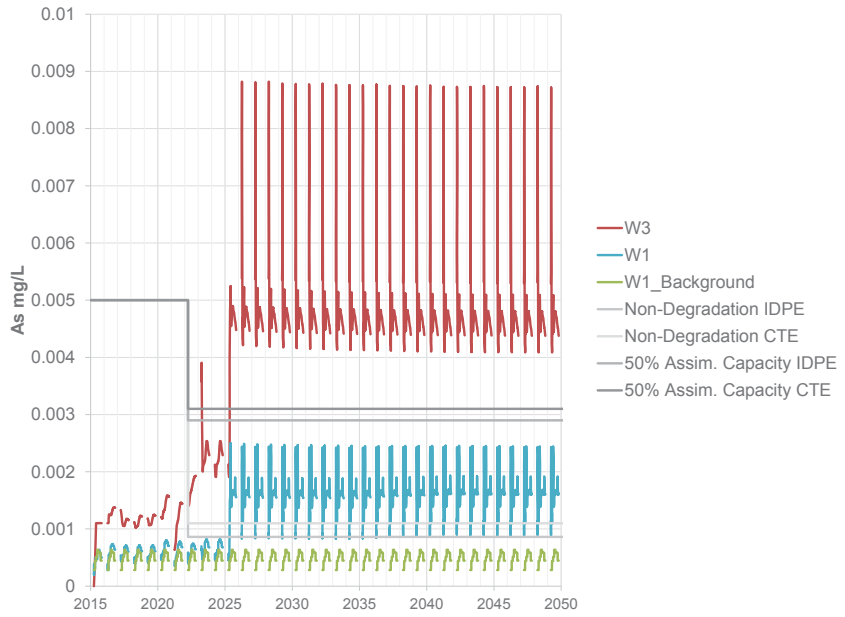


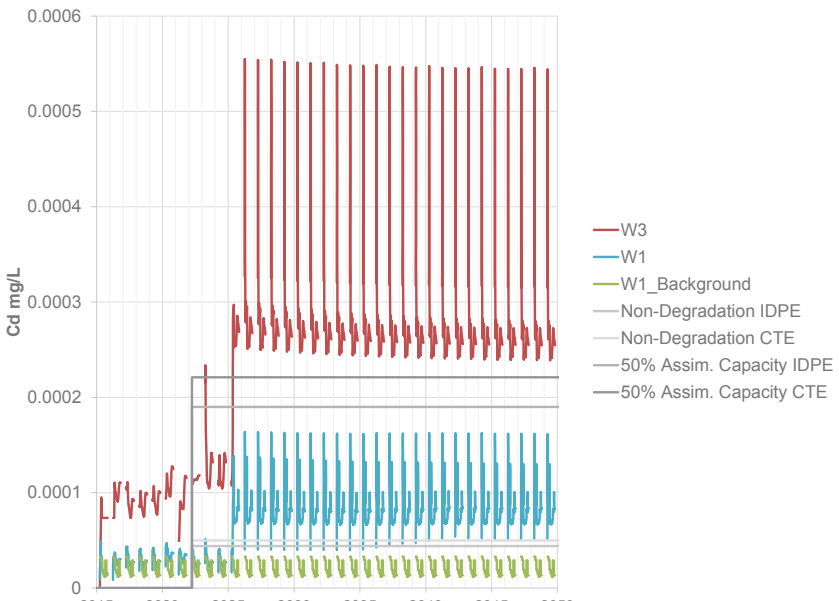
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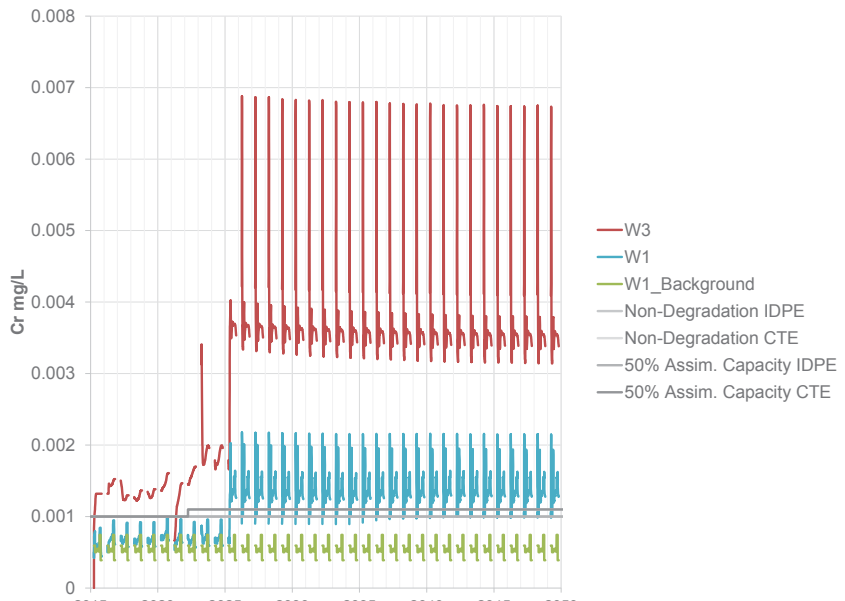


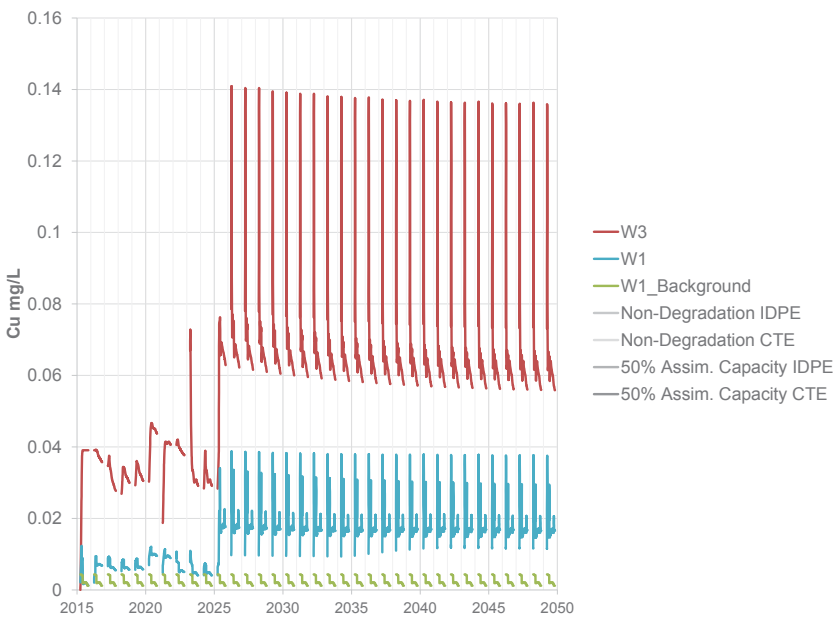


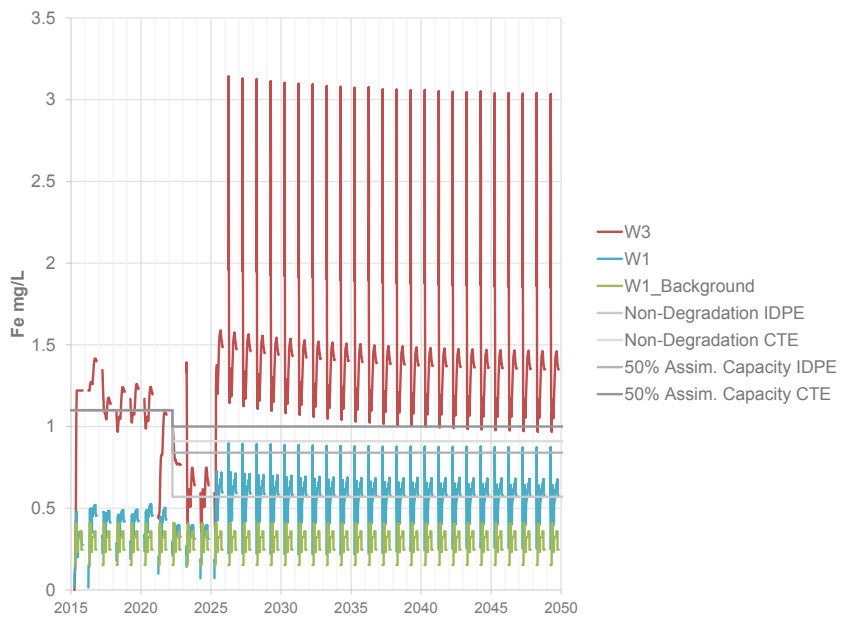




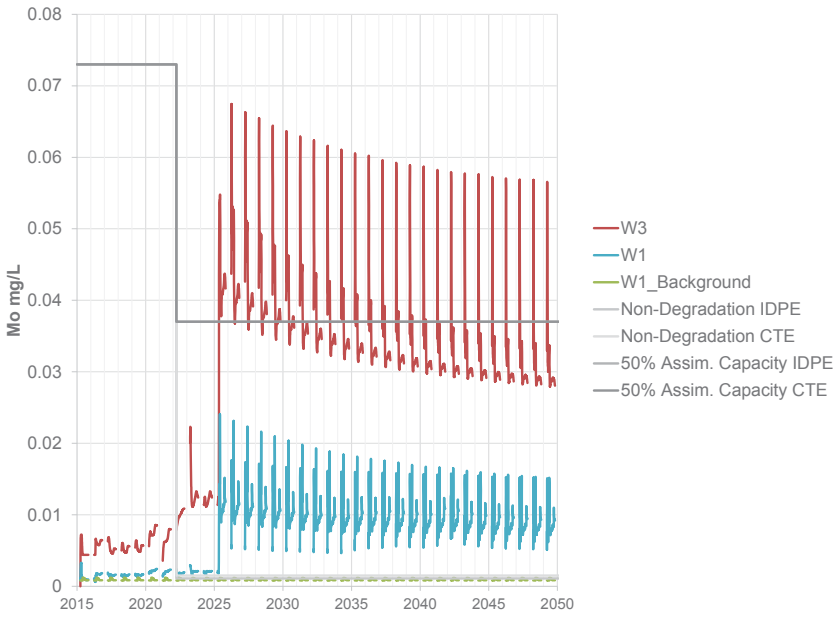


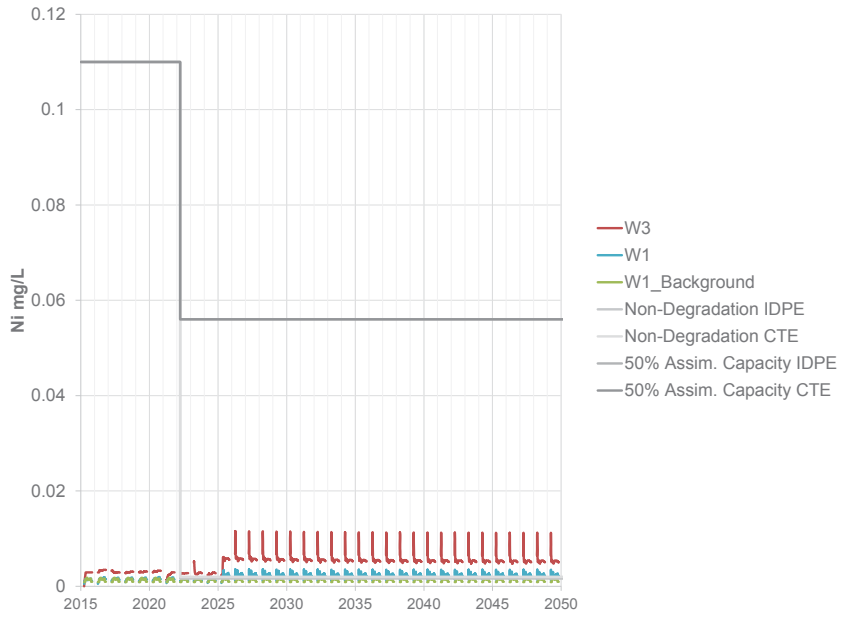


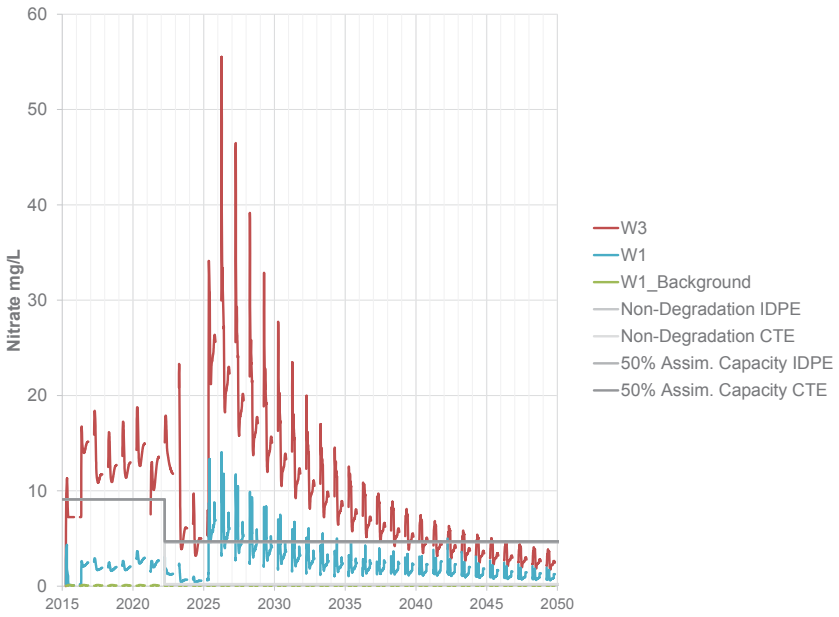


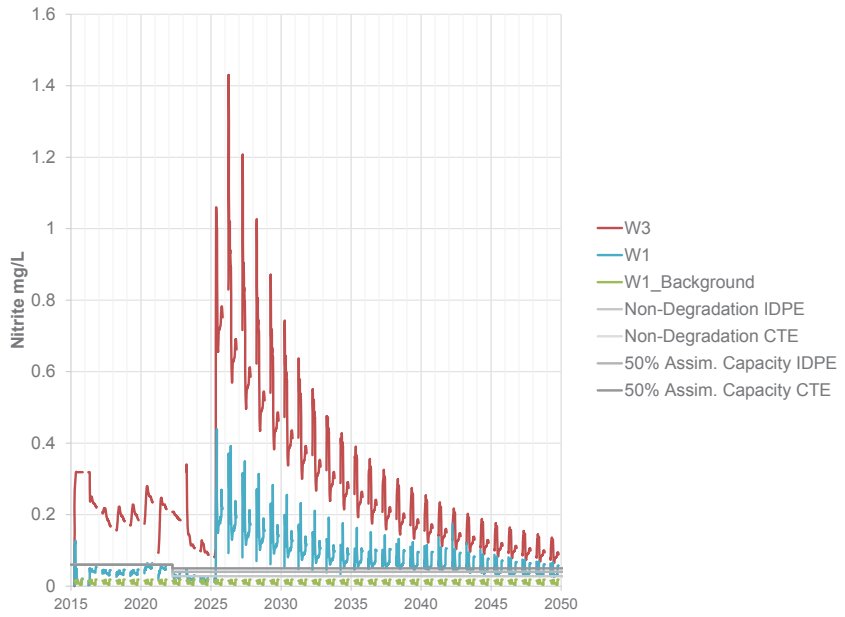


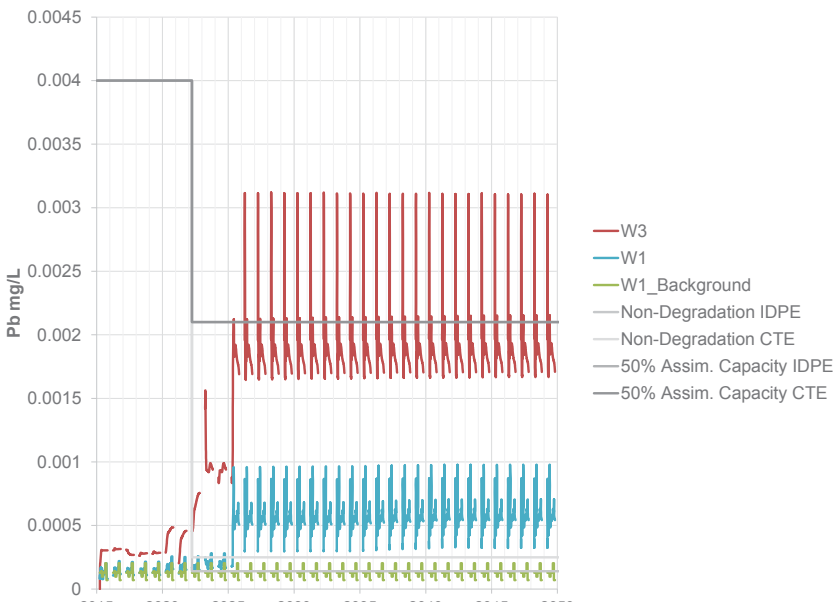
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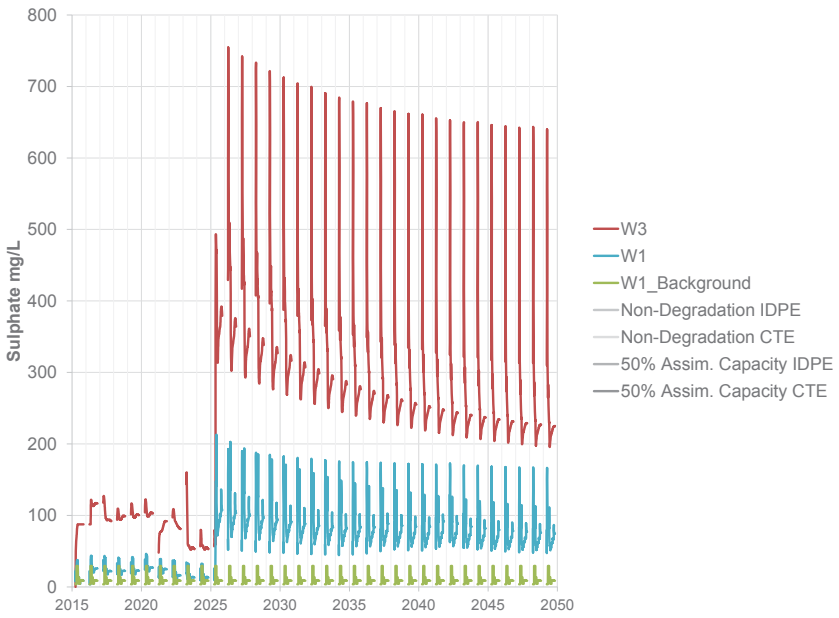


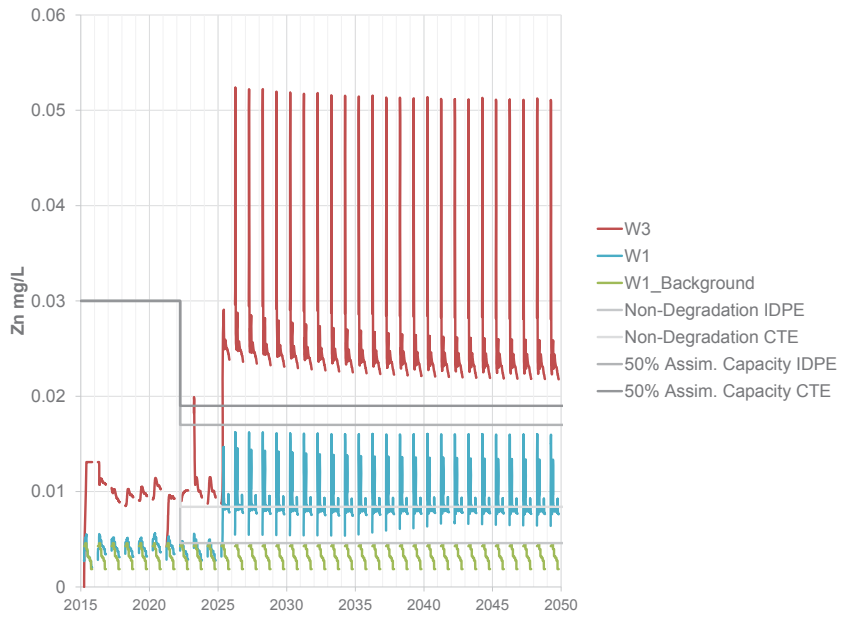




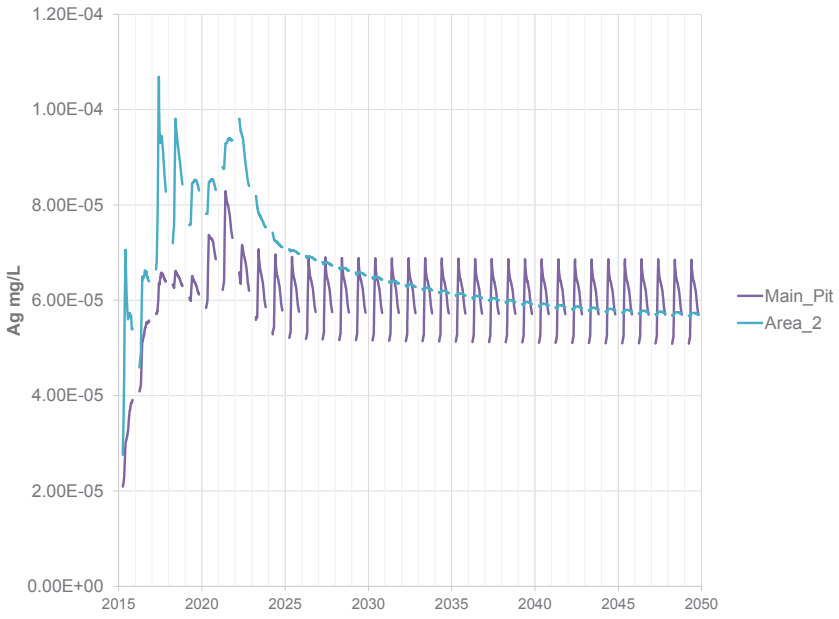


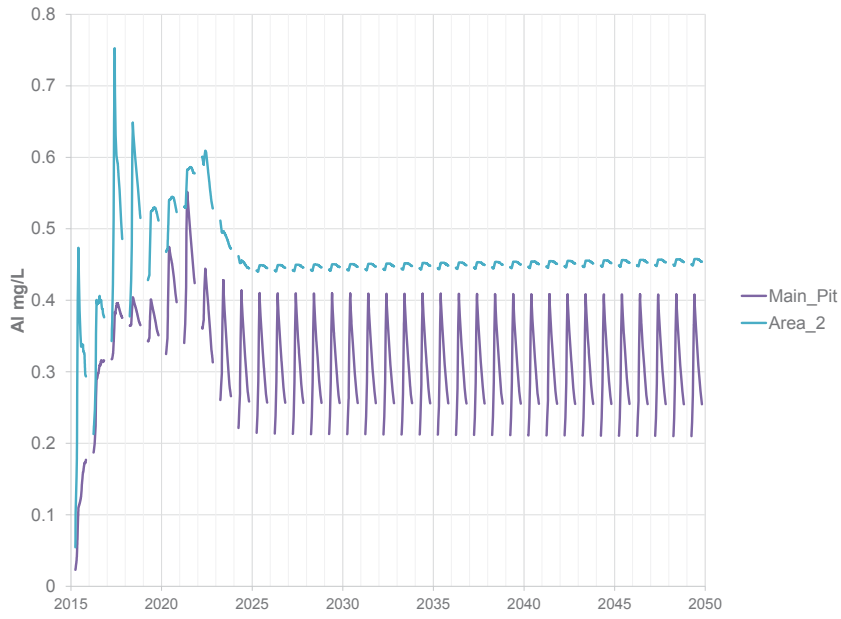


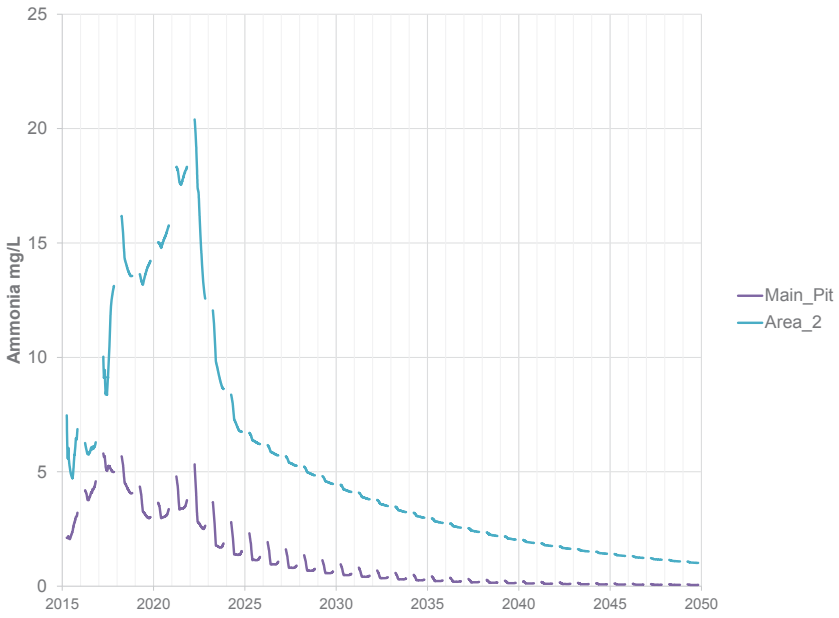


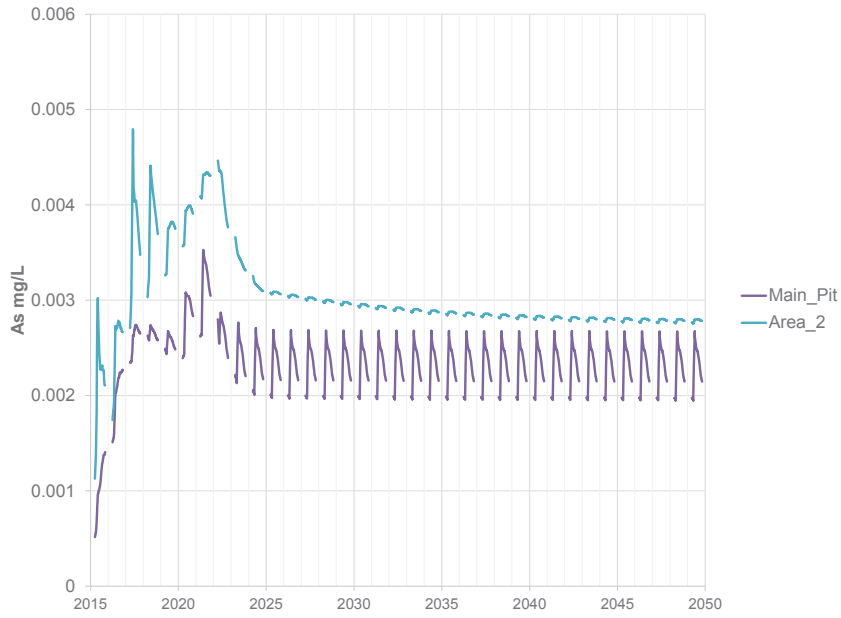


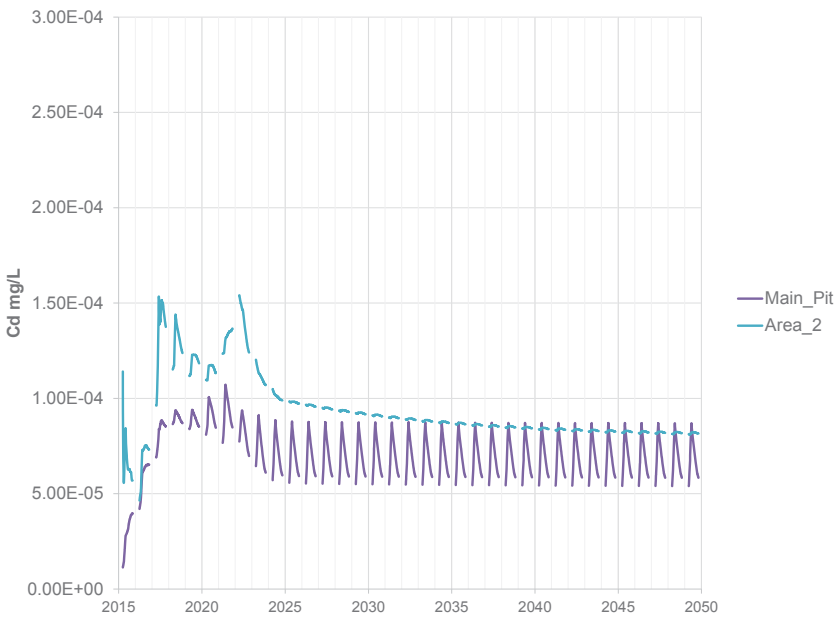
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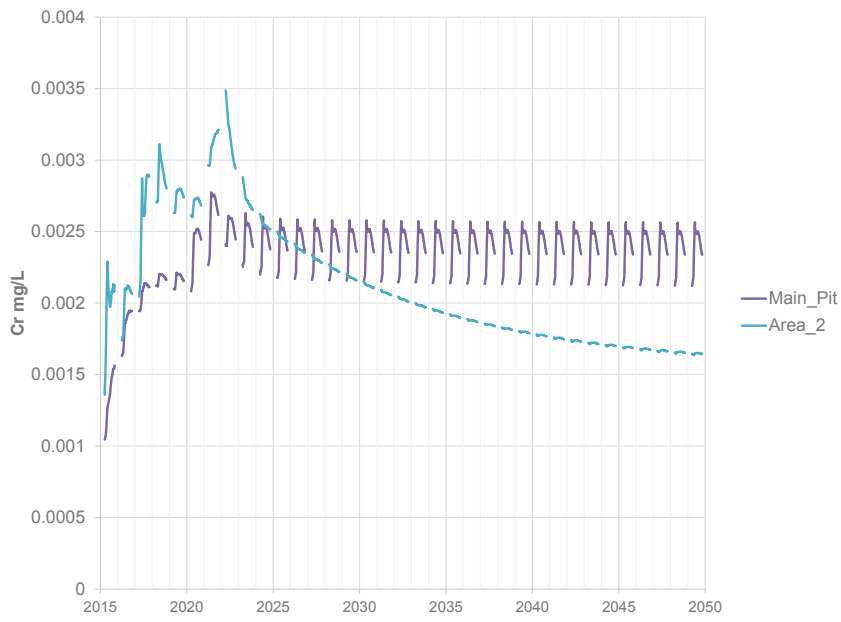


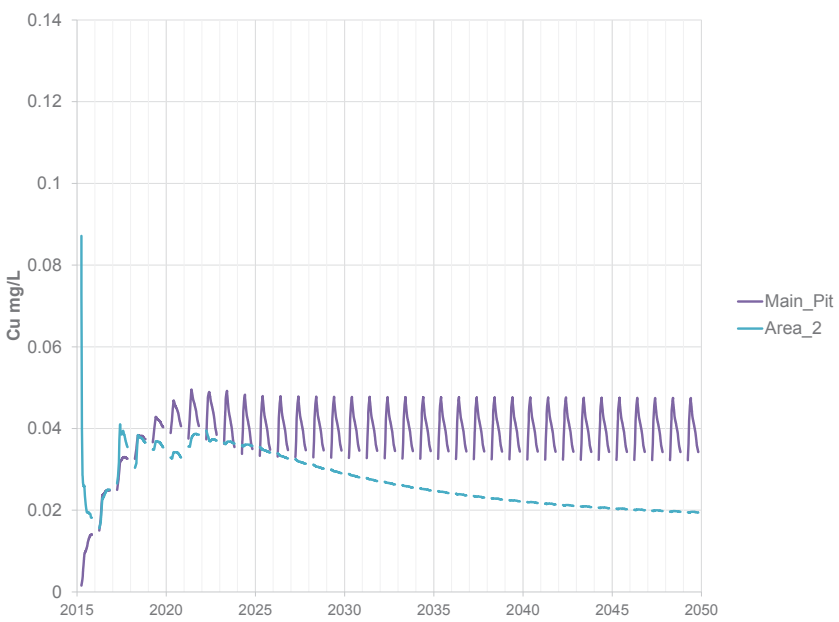


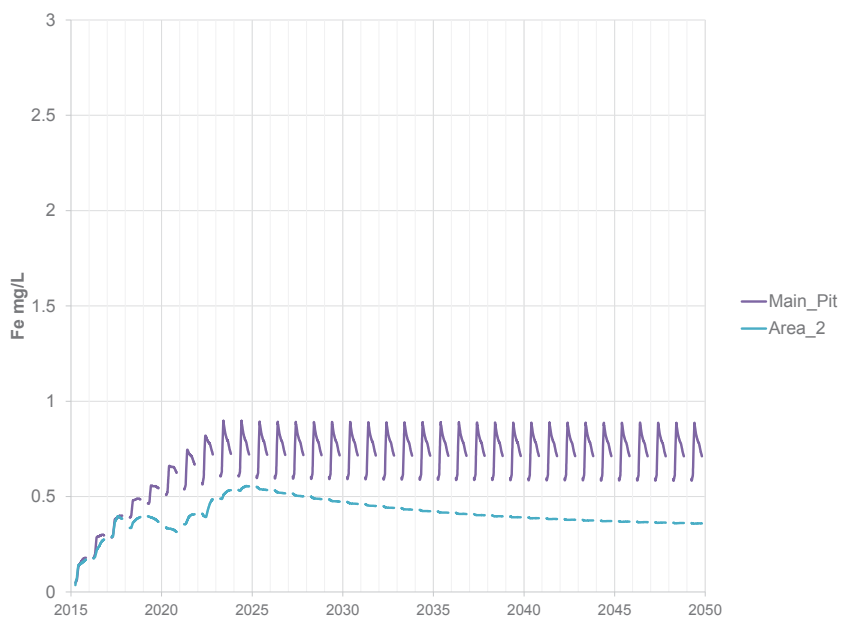




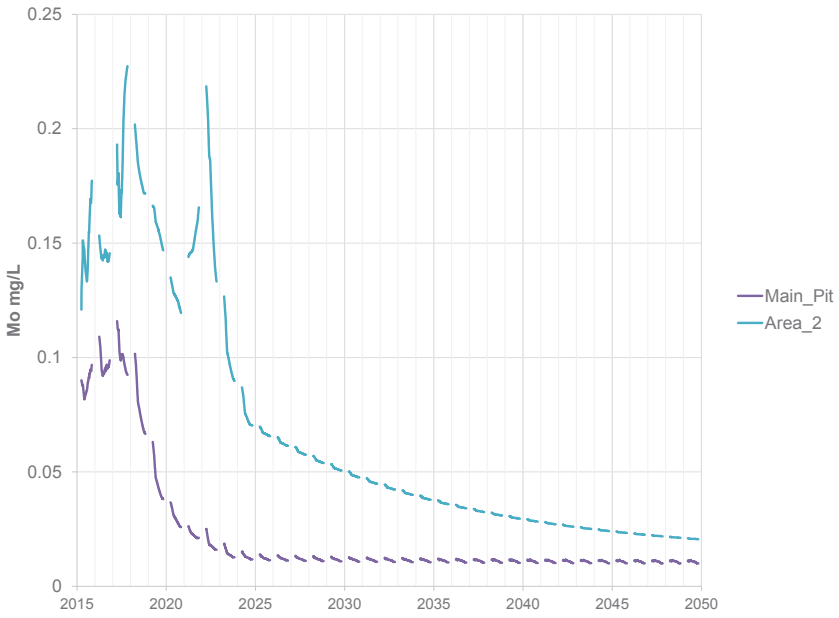


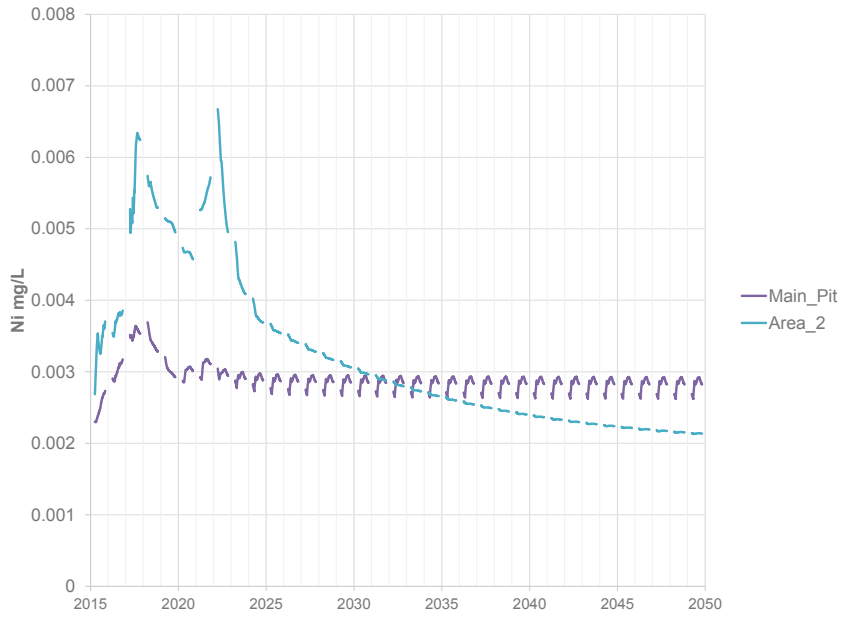


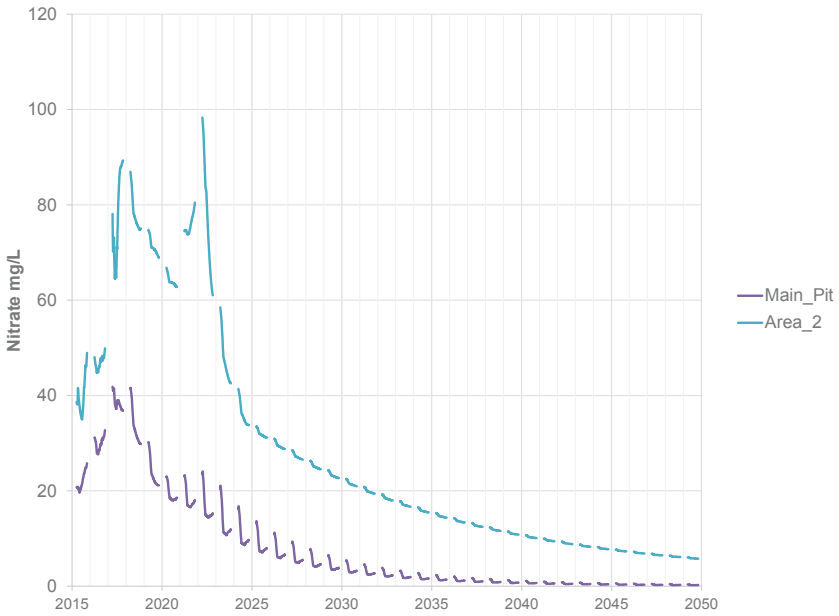


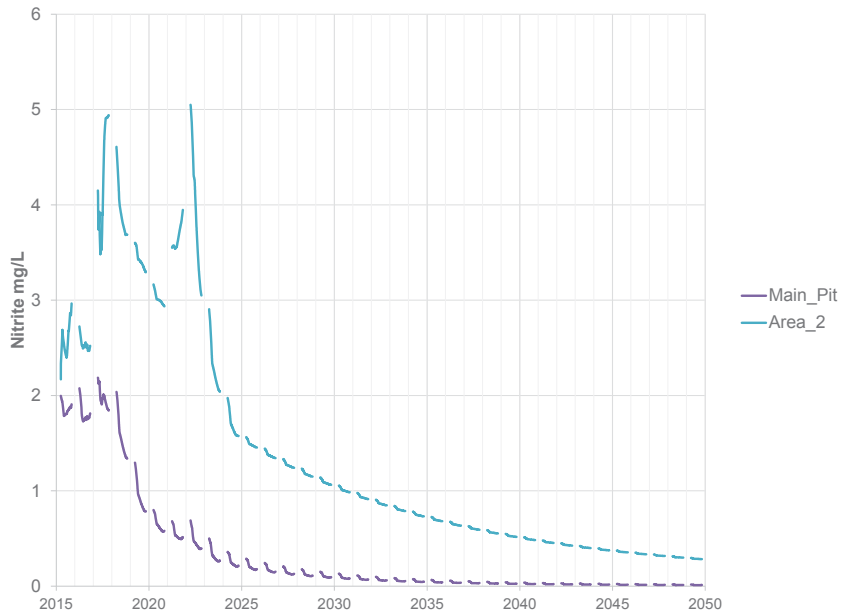


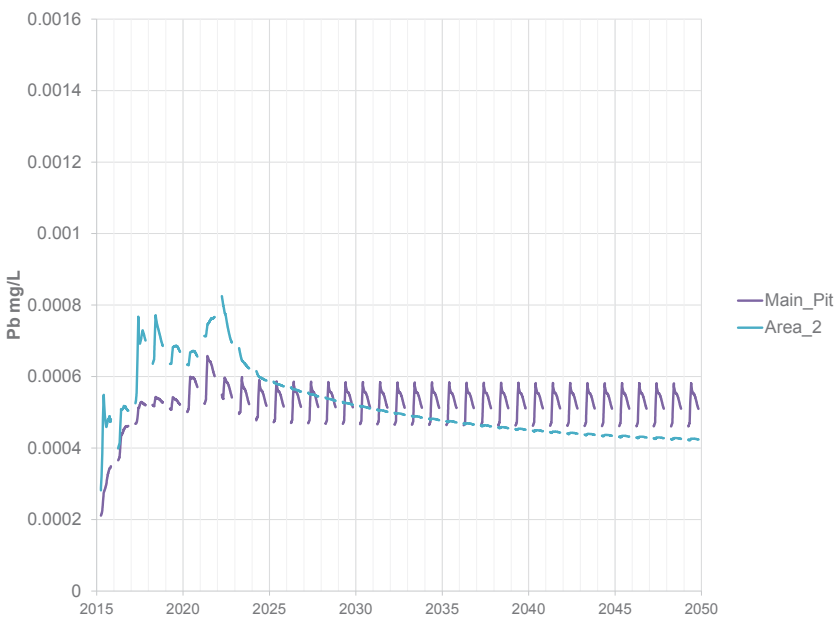
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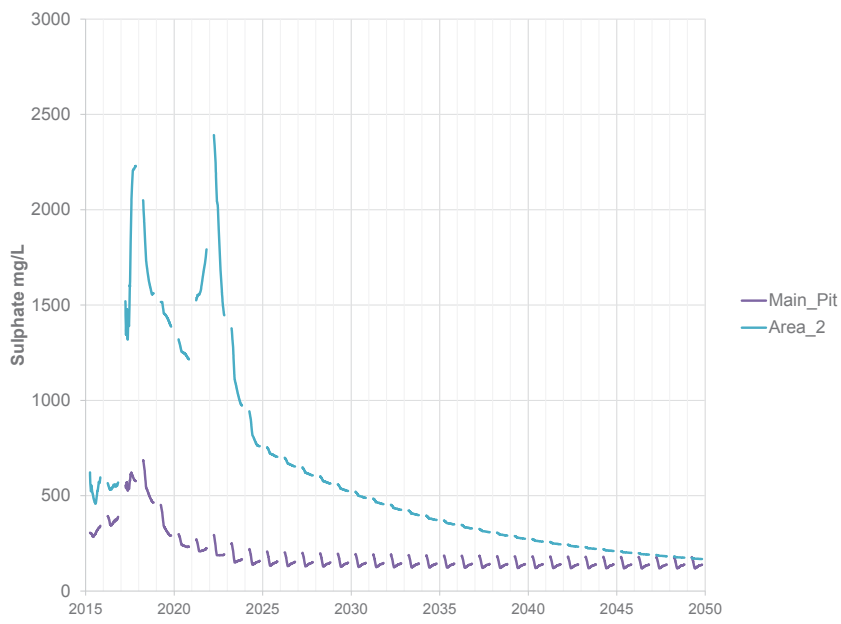


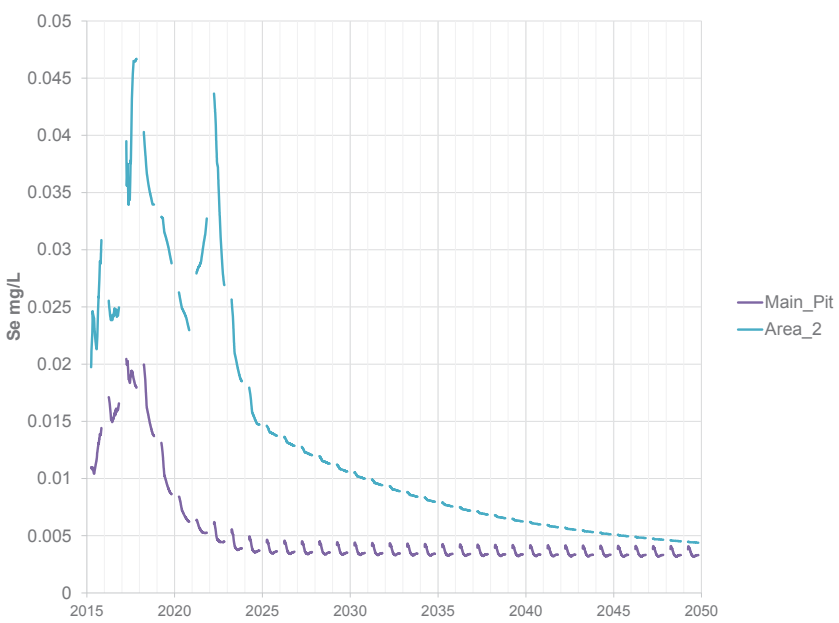


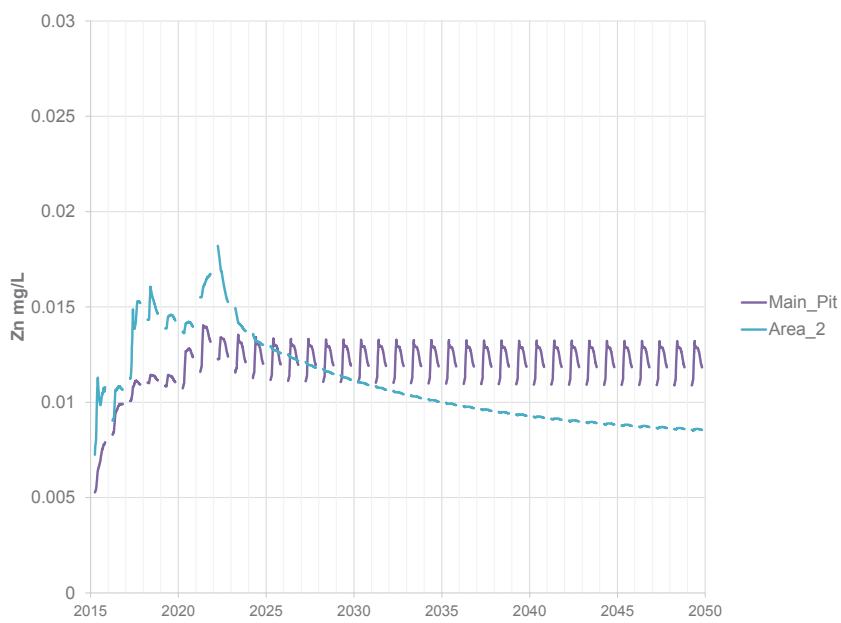




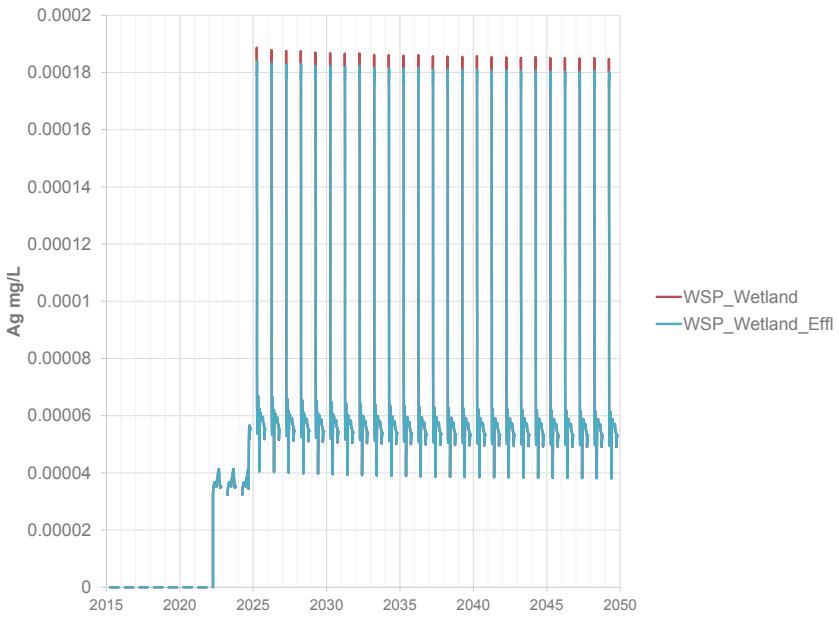


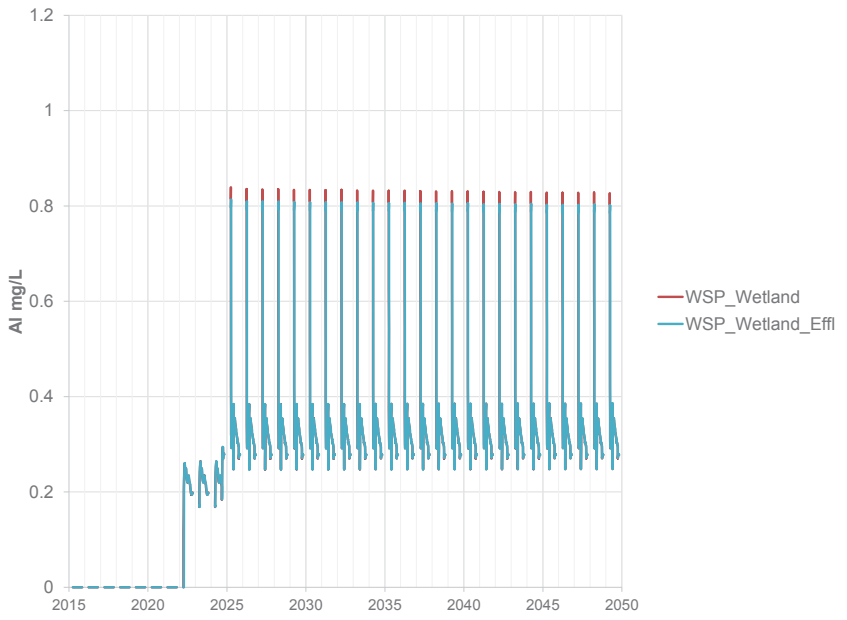


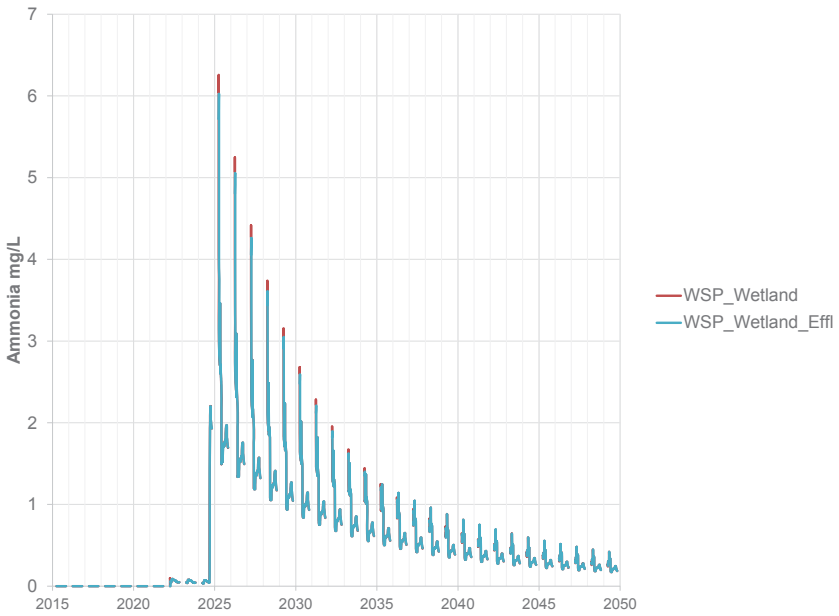


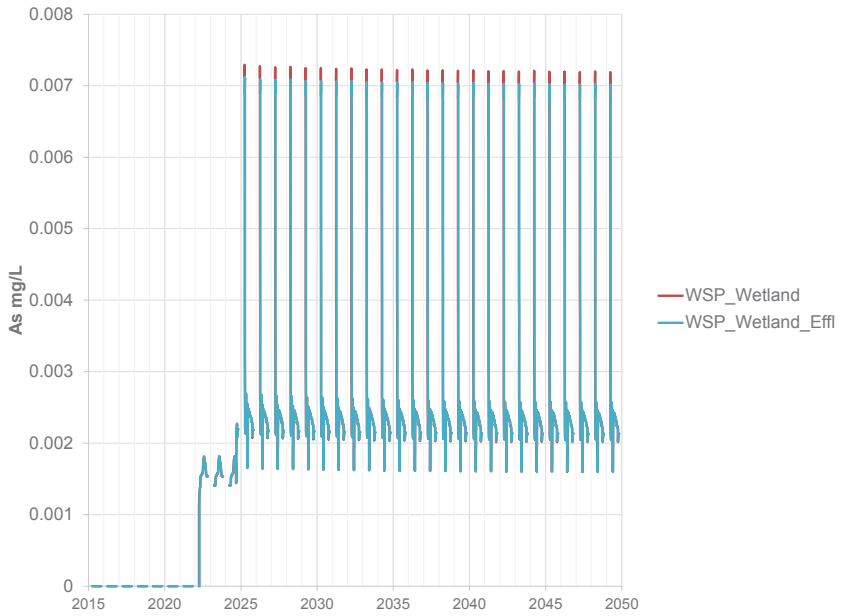


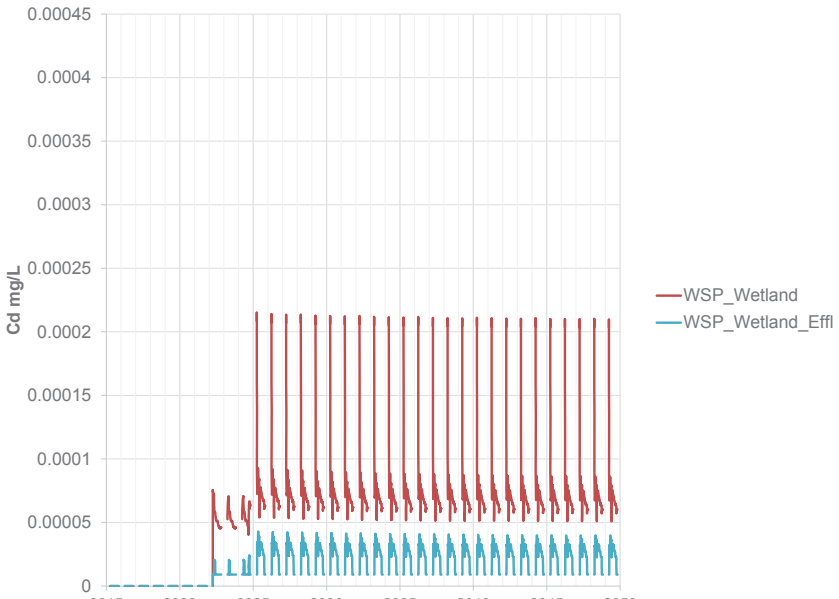
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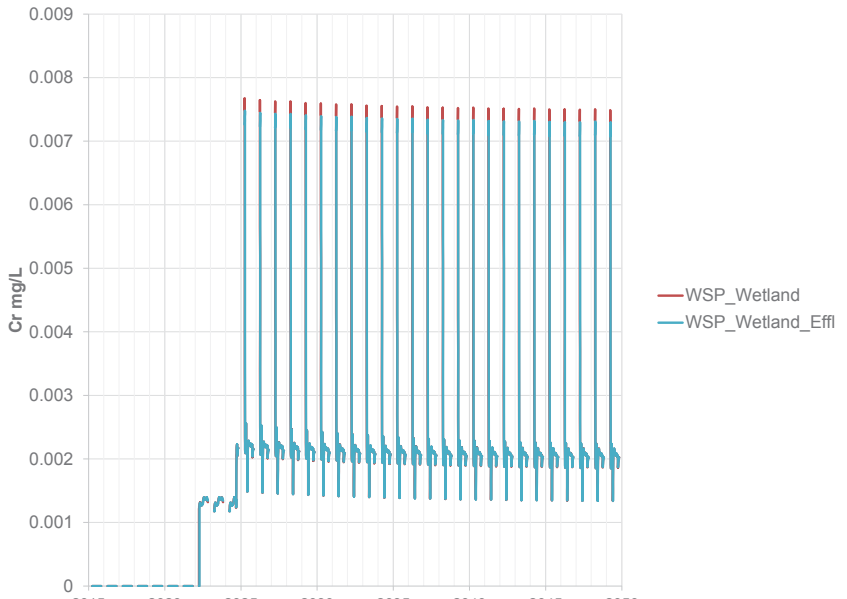


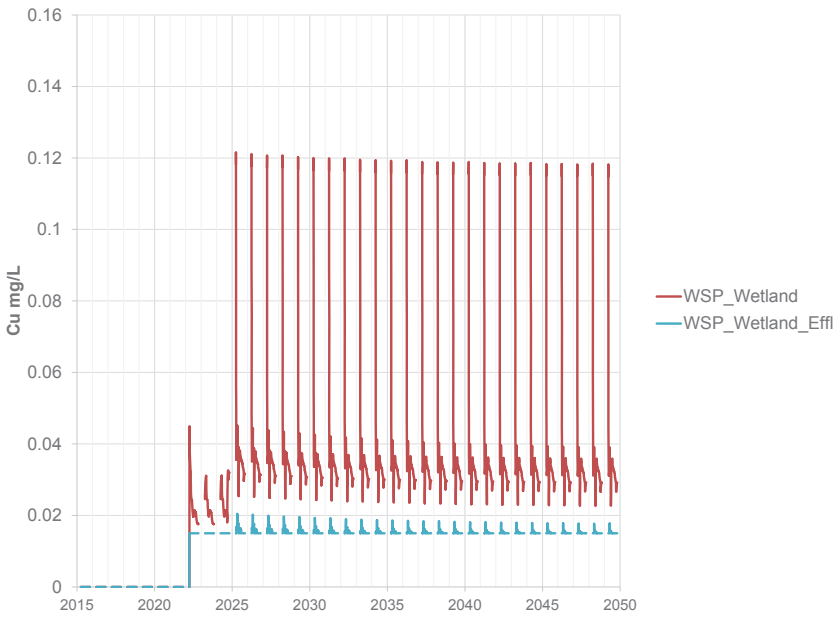


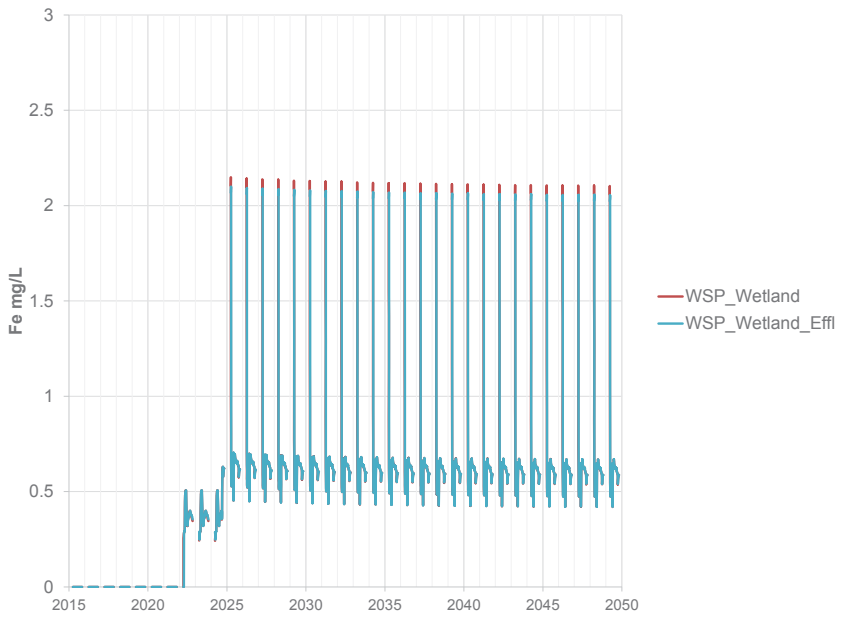




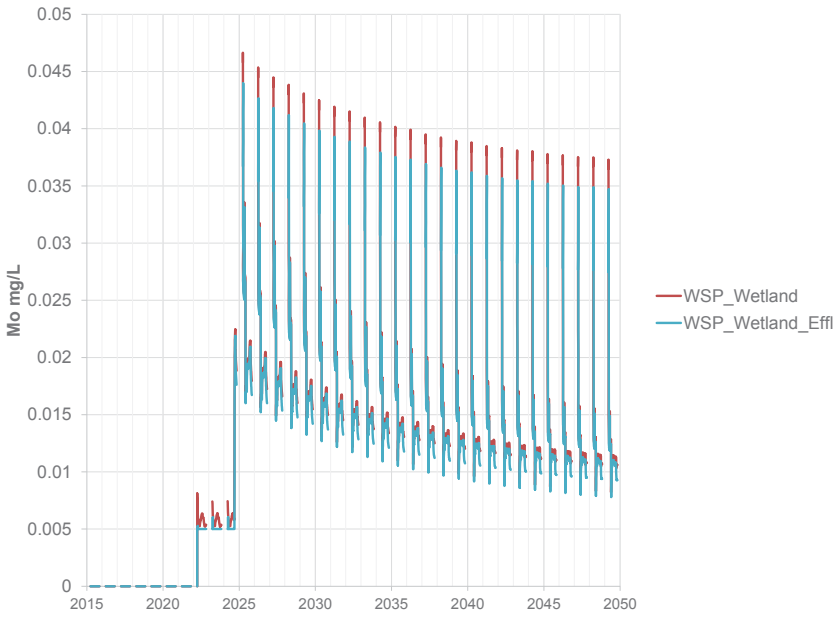


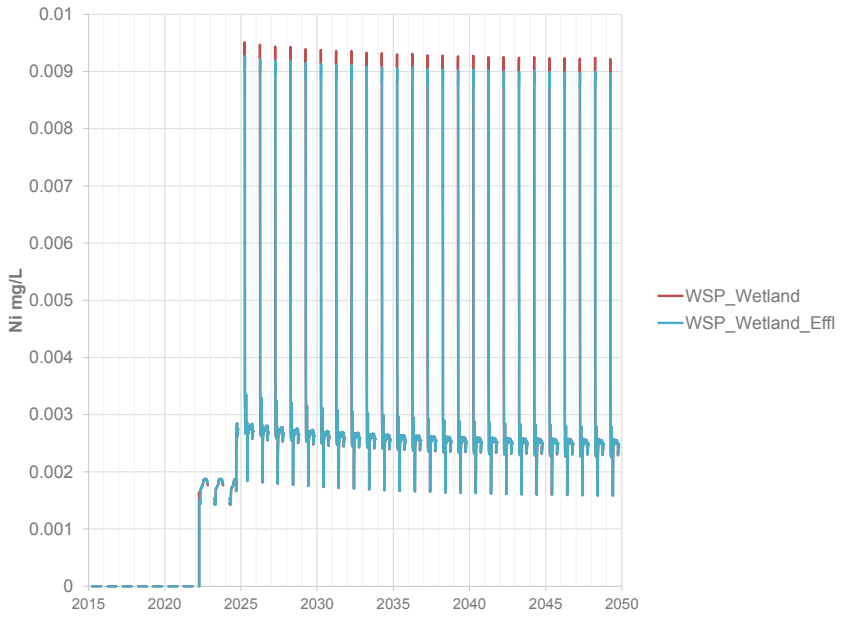


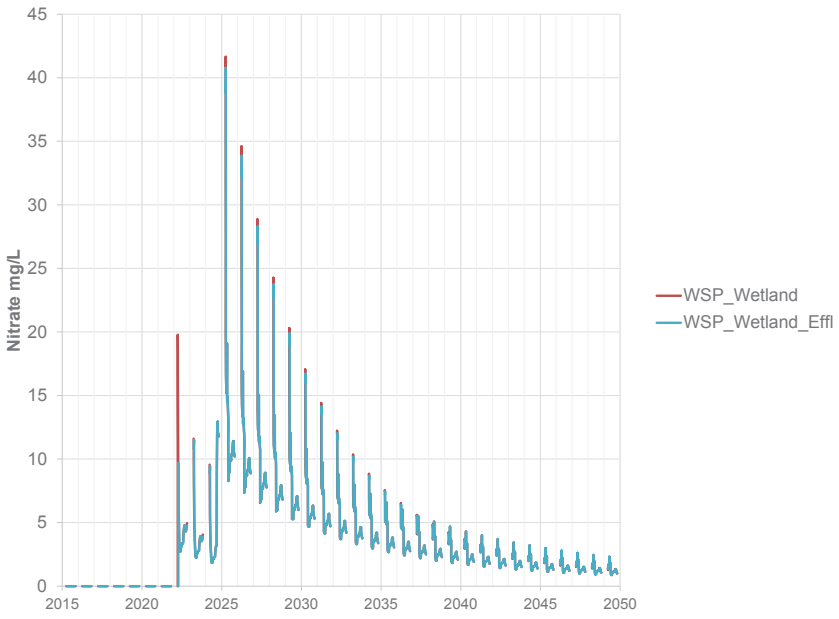


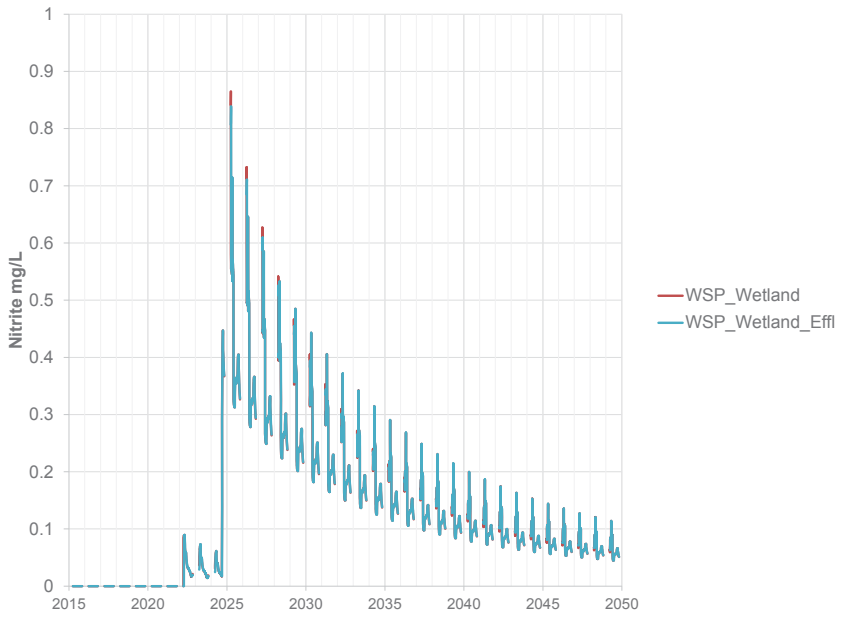


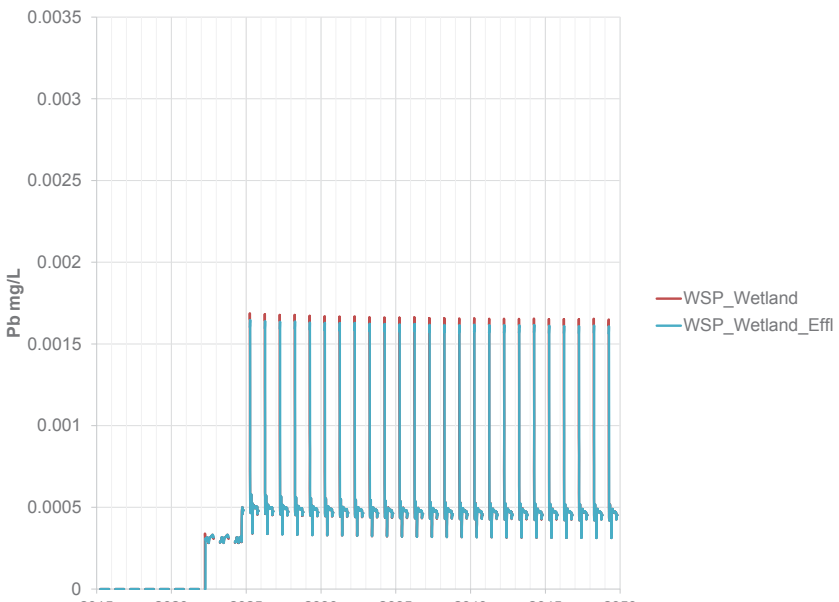
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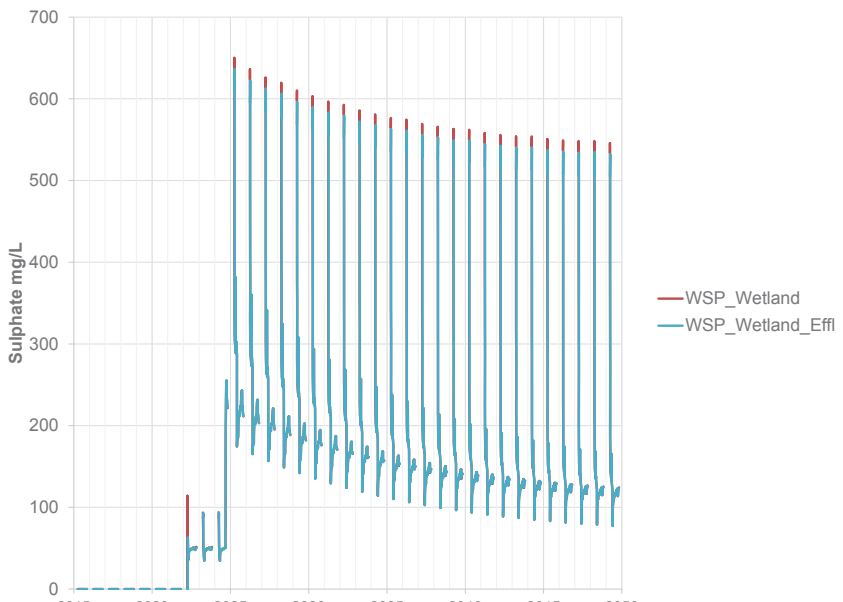


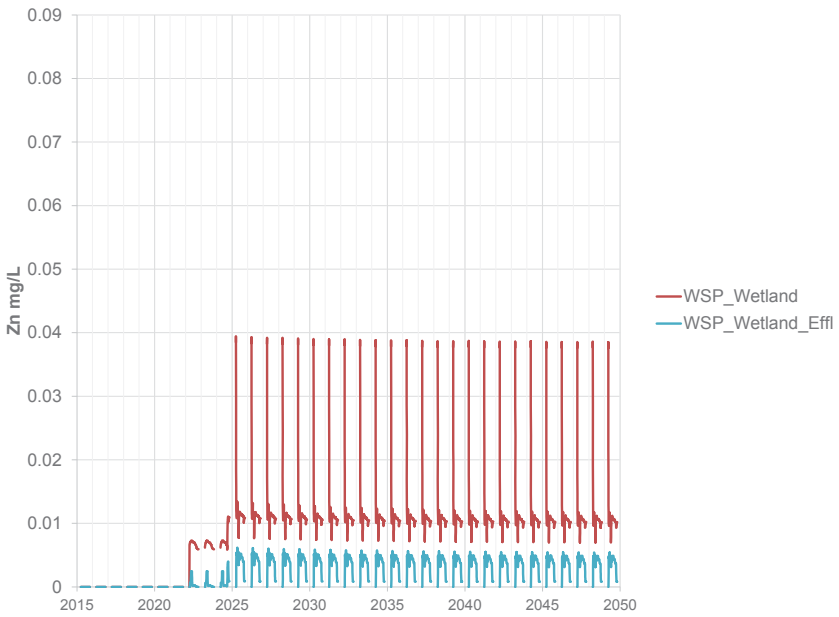


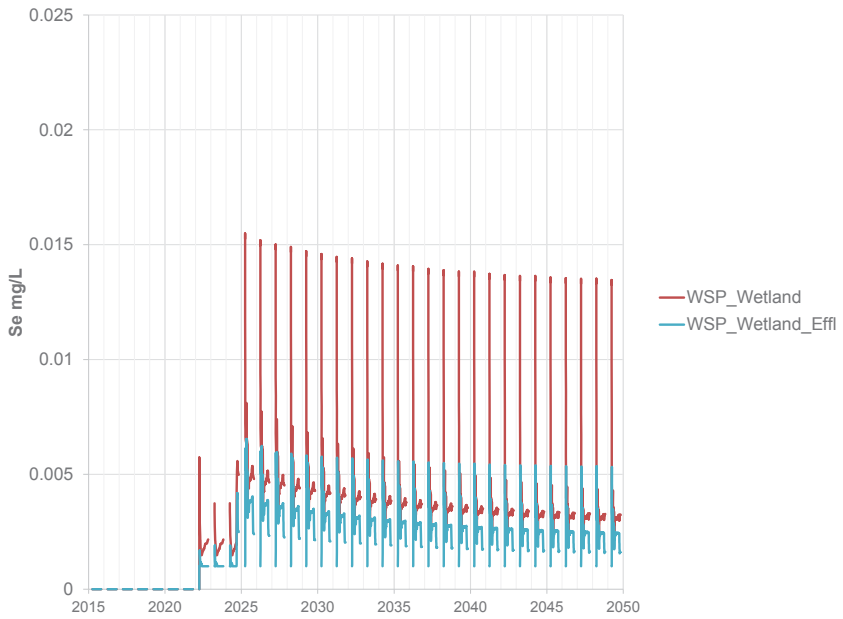




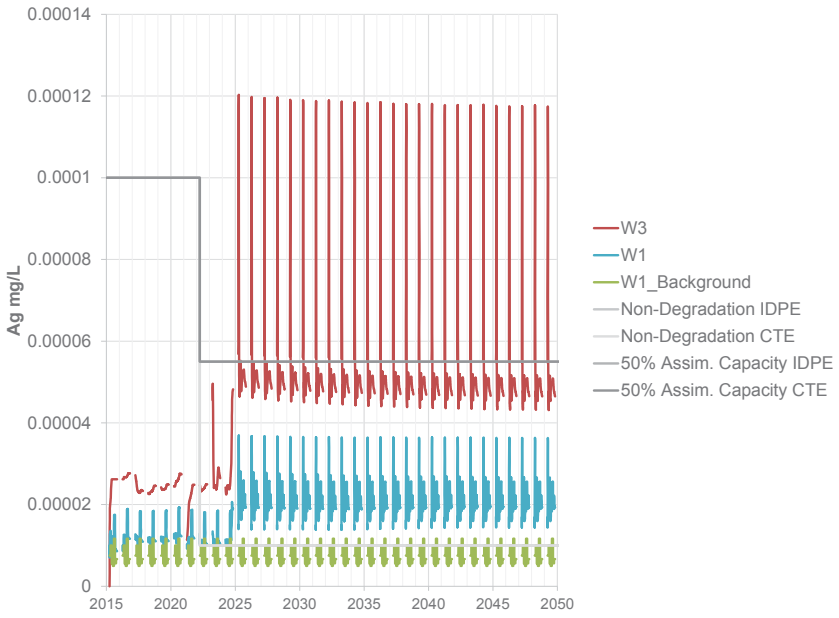


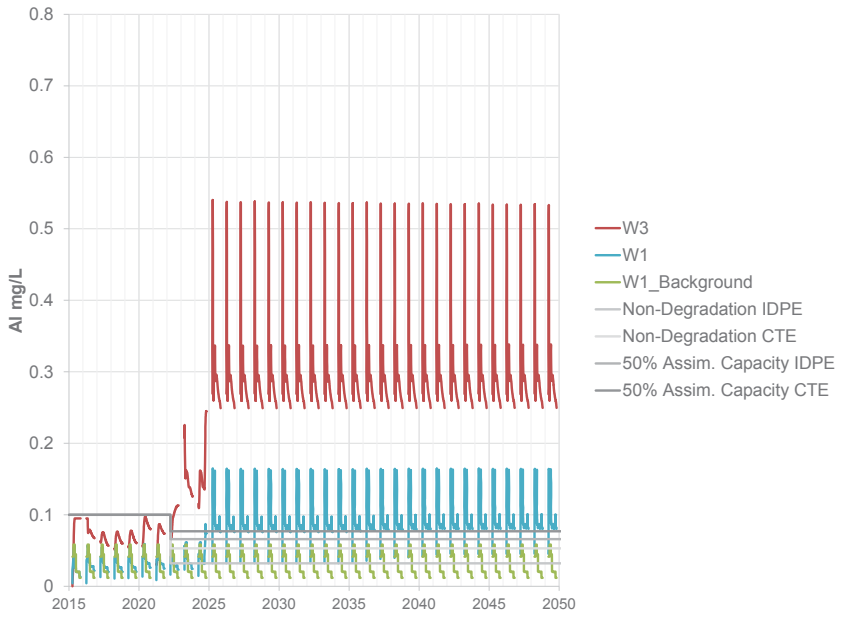


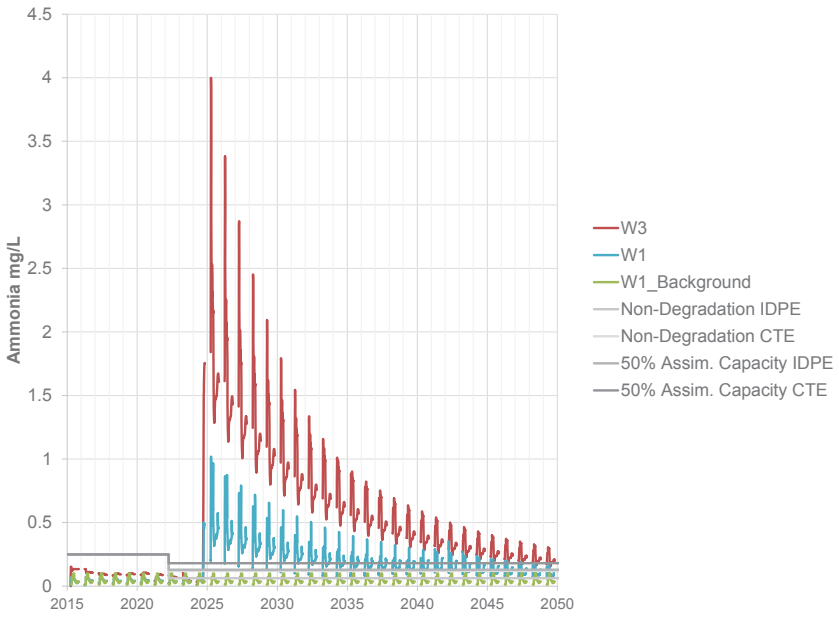


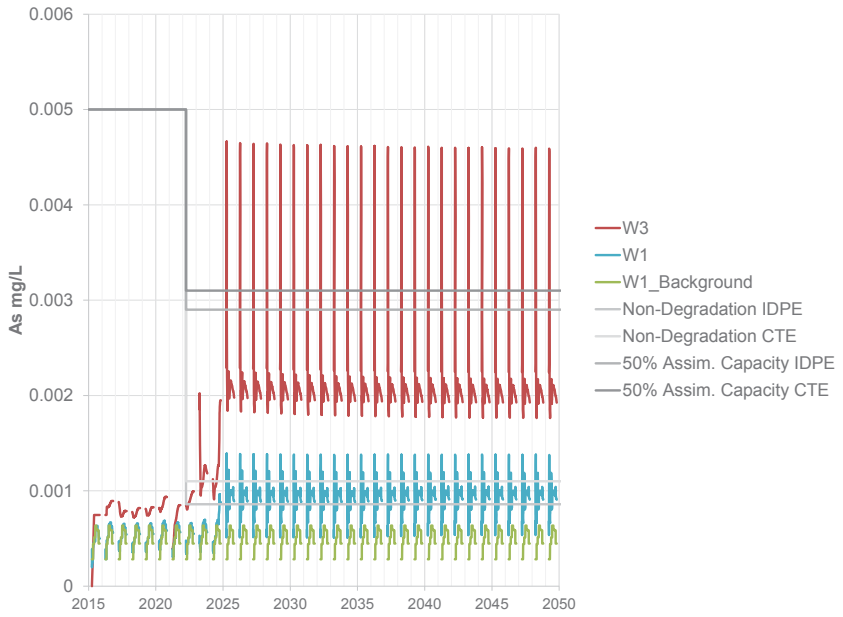


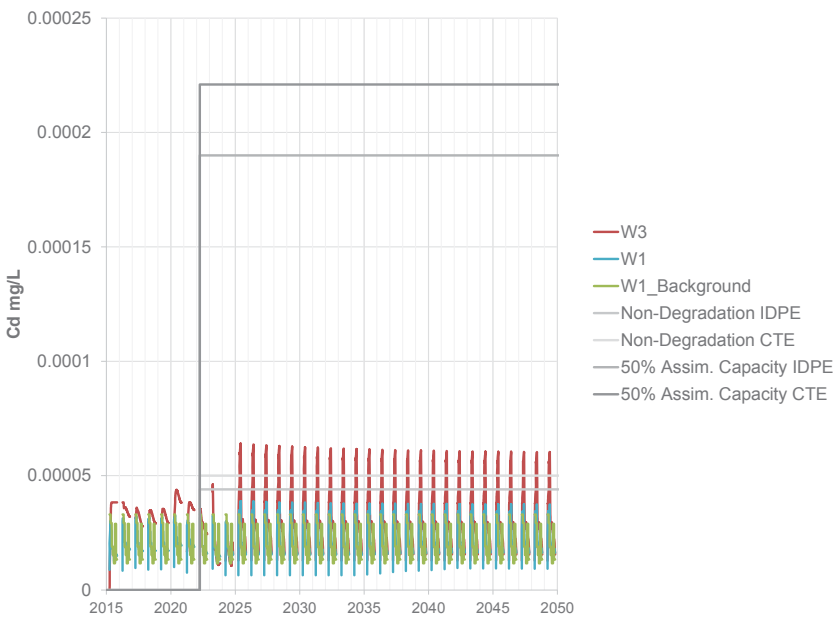
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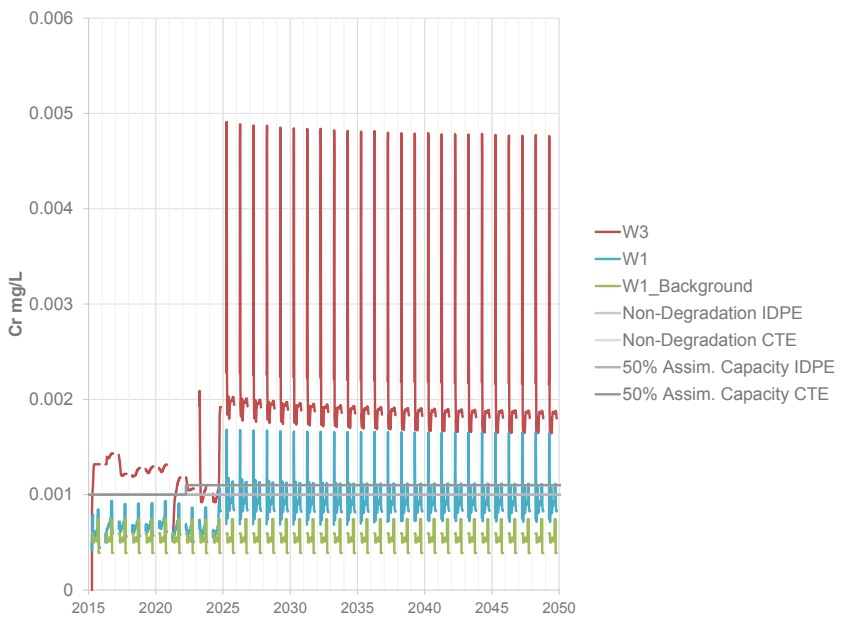


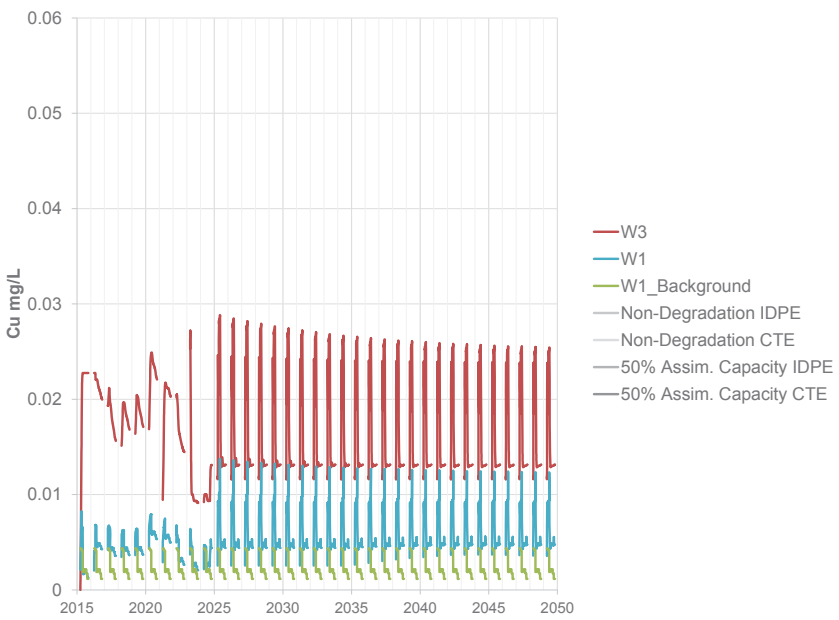


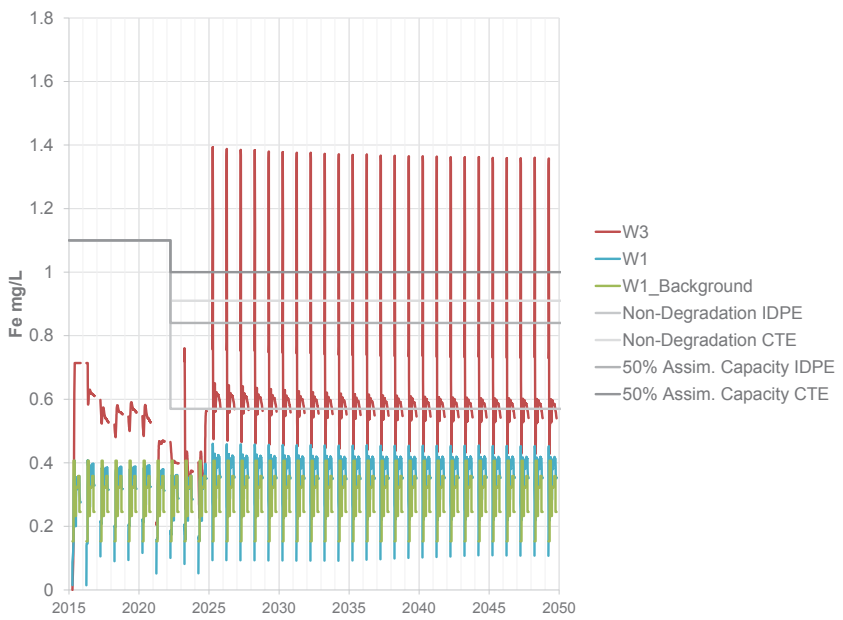




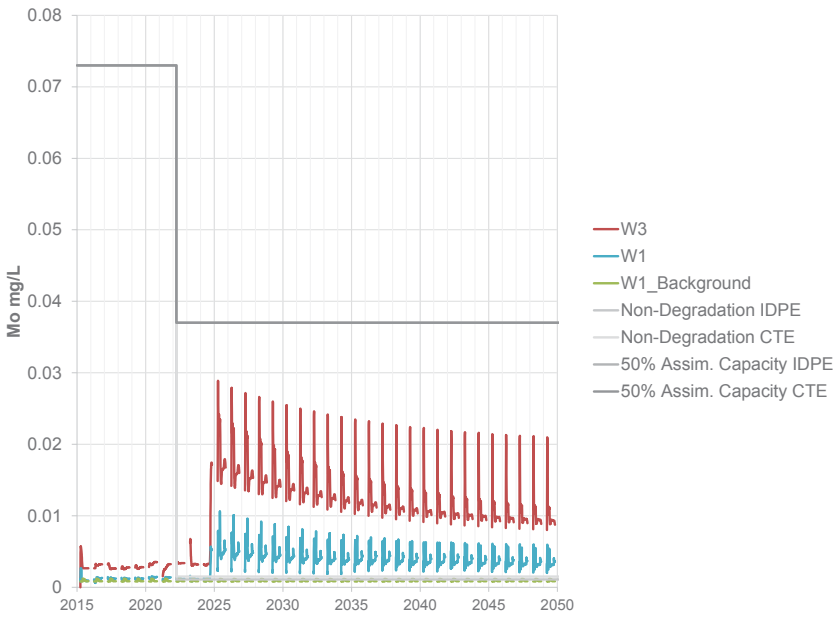


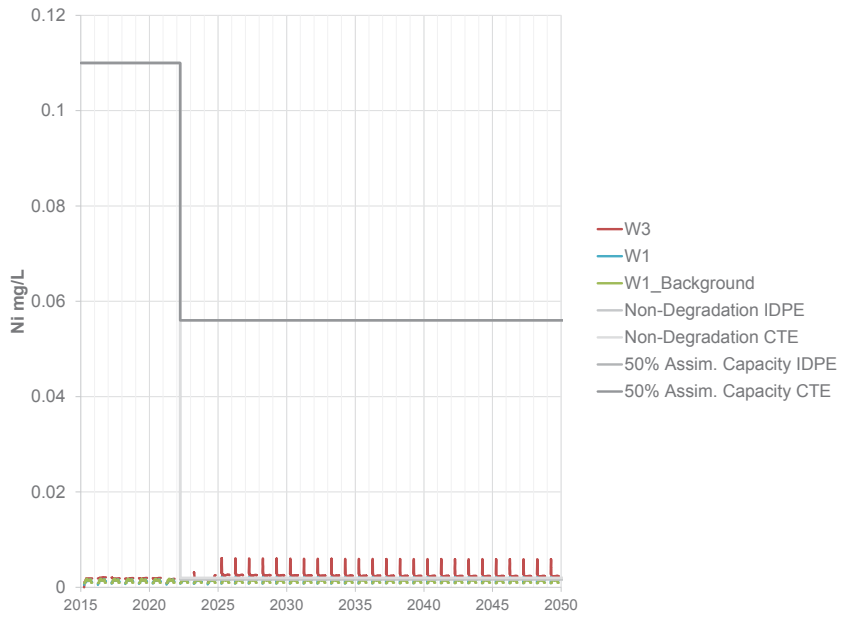


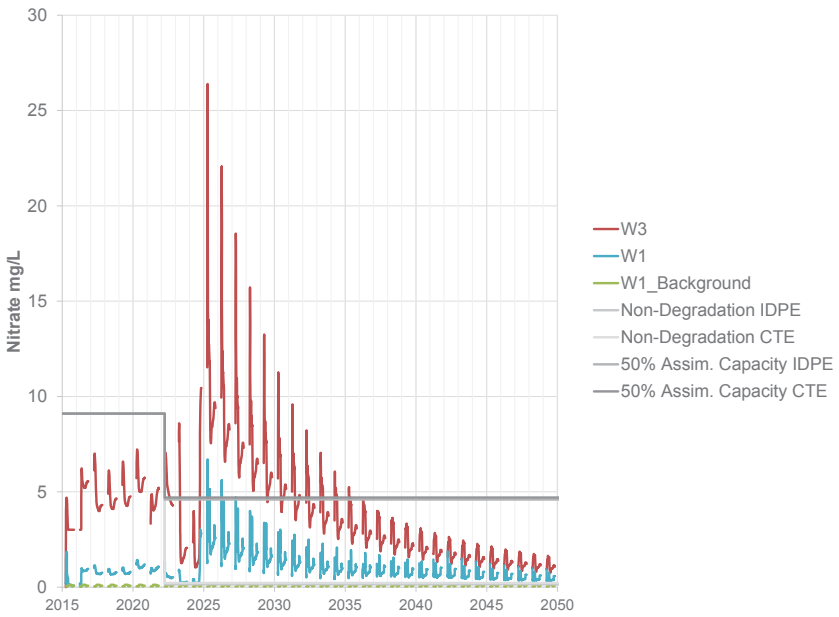


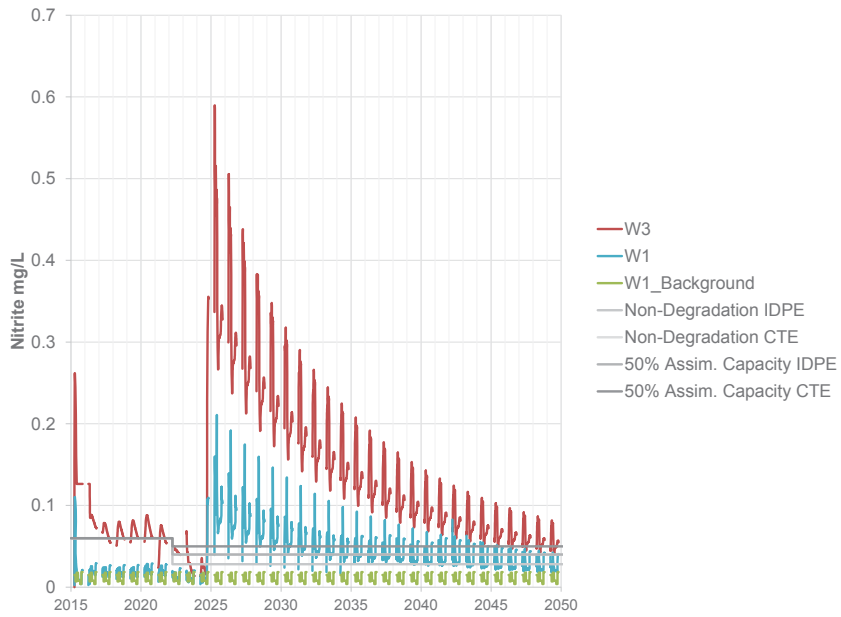


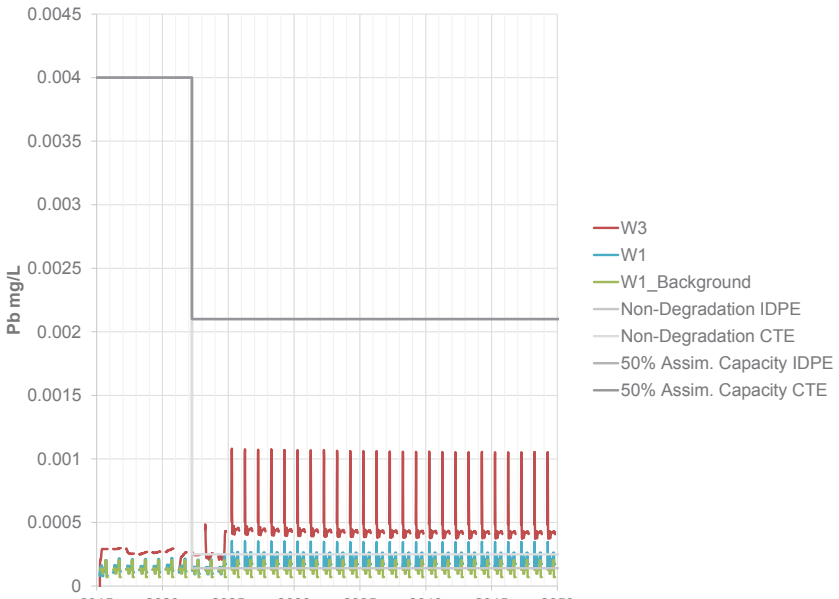
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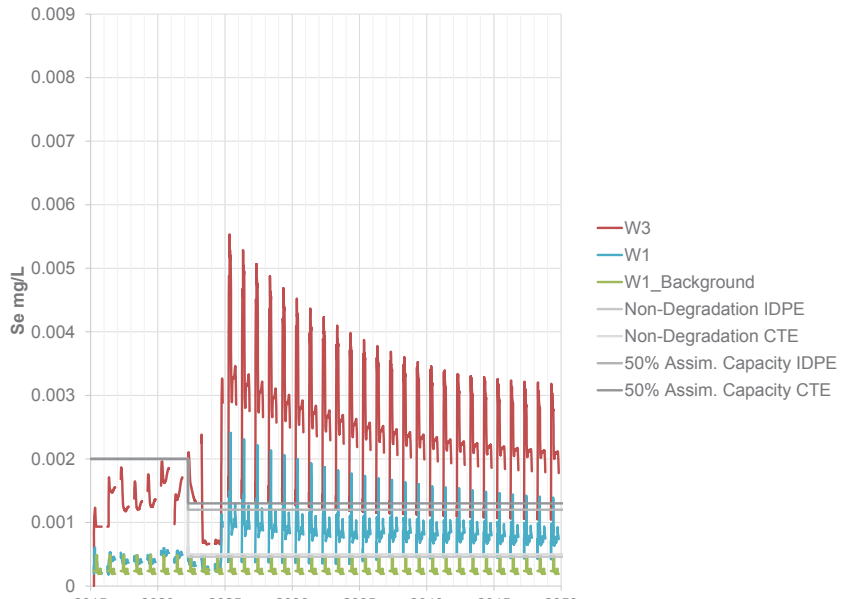


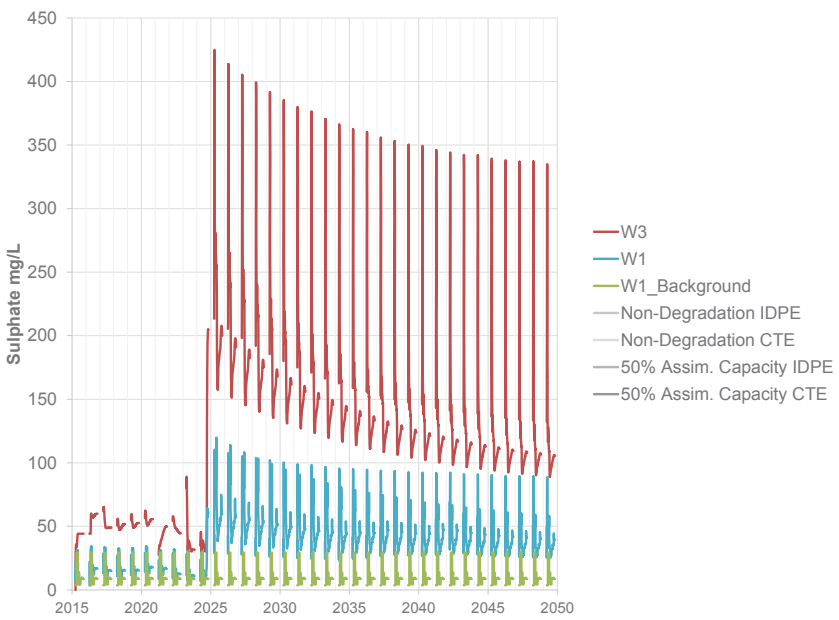


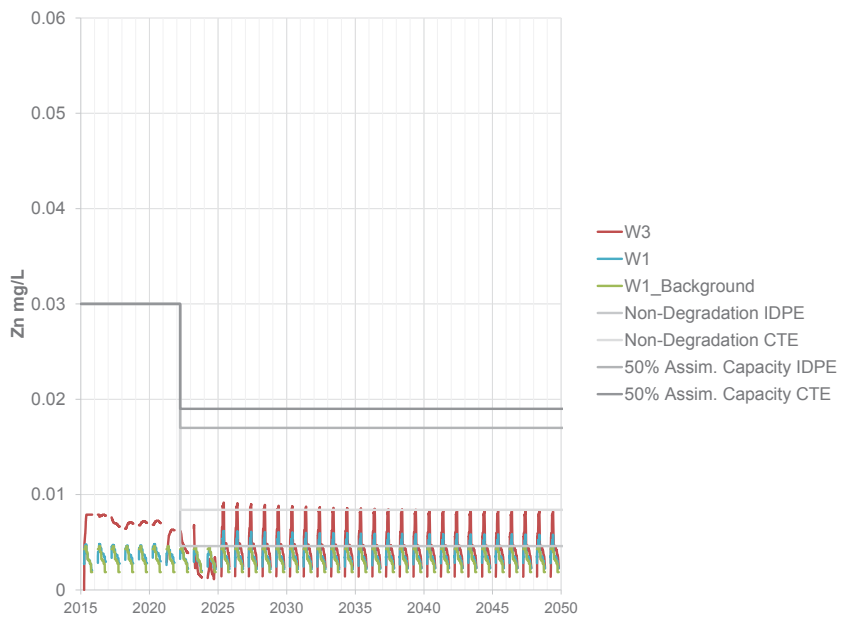




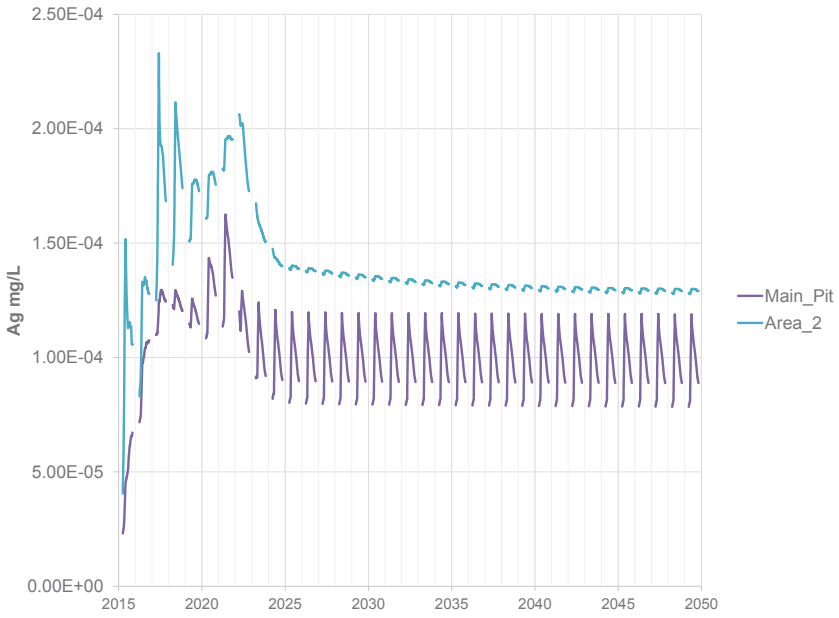


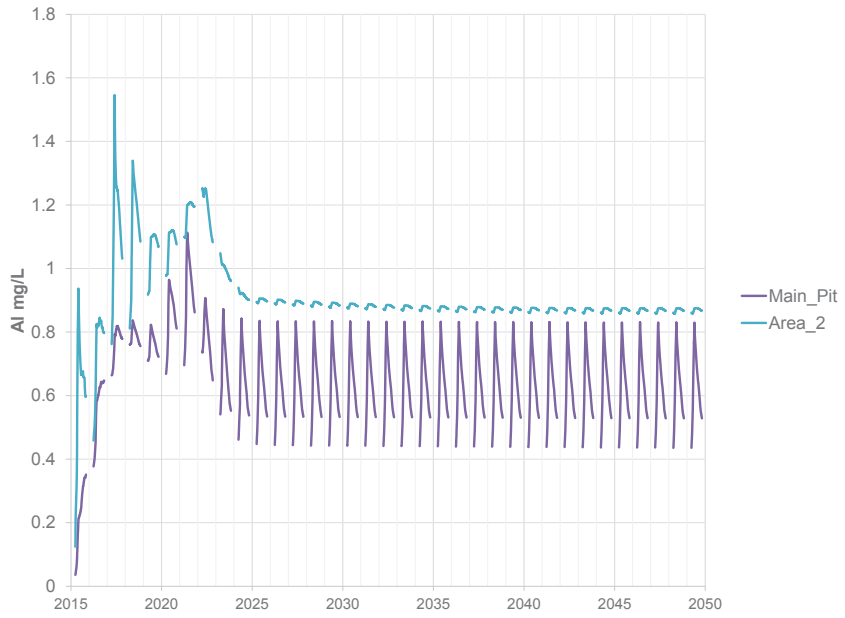


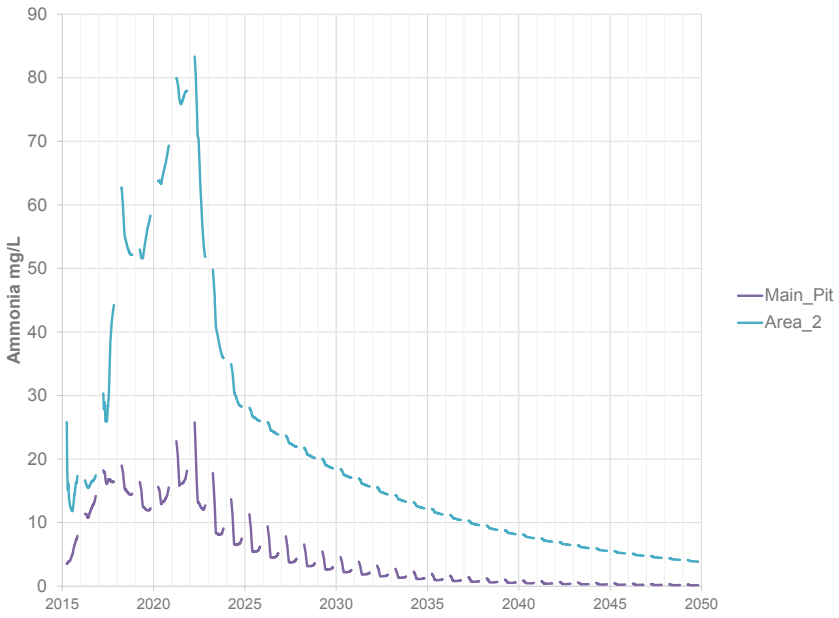


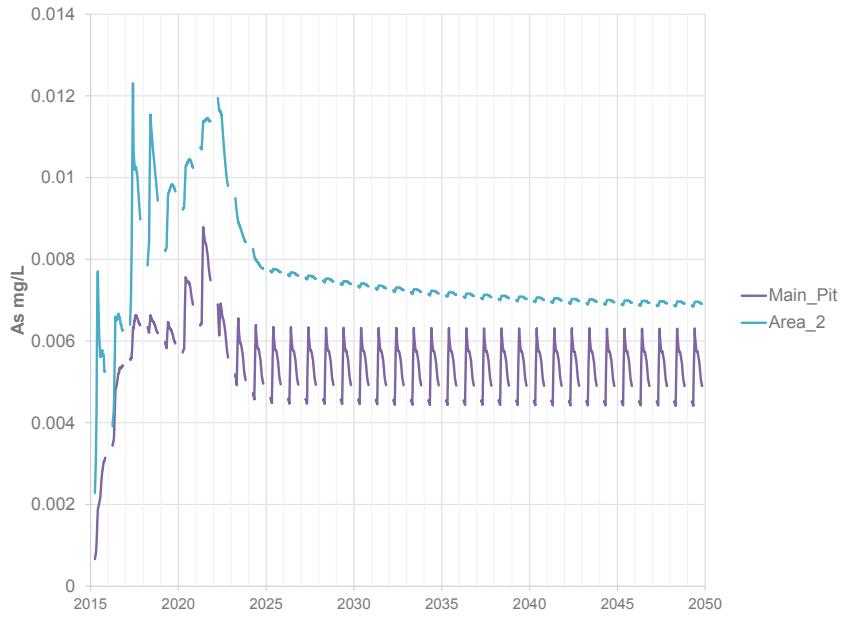


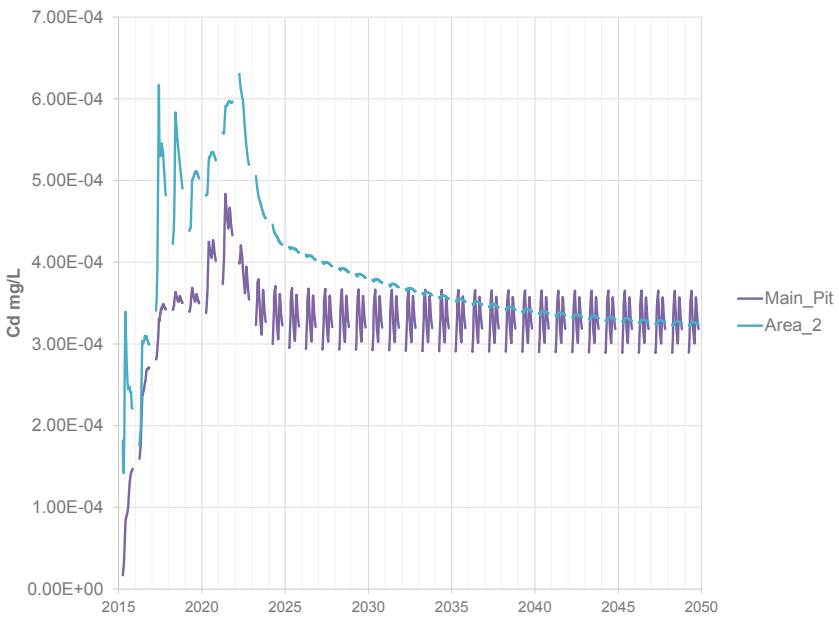
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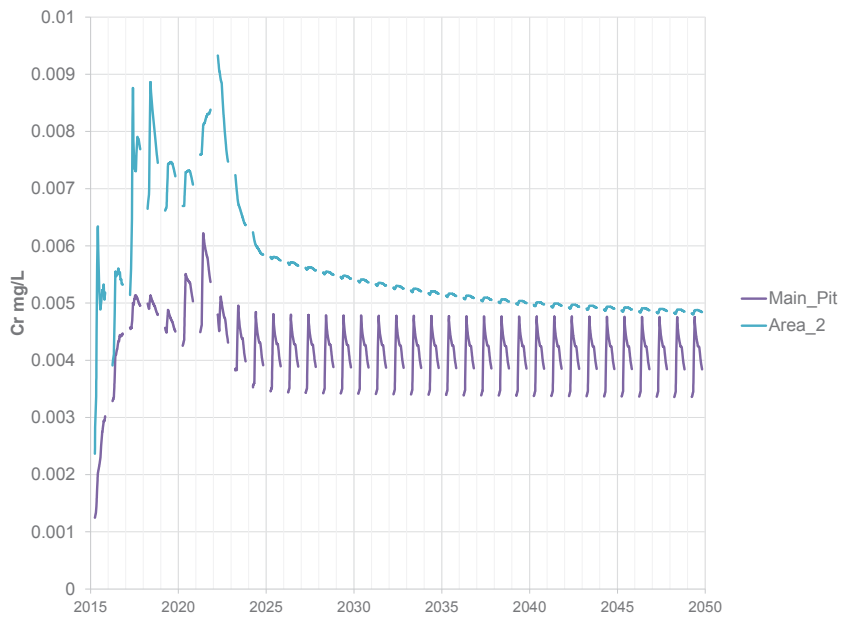

Ag mg/L
2.50E-04
2.00E-04
1.50E-04
1.00E-04
5.00E-05
0.00E+00
2015
2020
2025
2030
2035
2040
2045
2050
Main_Pit
Area_2

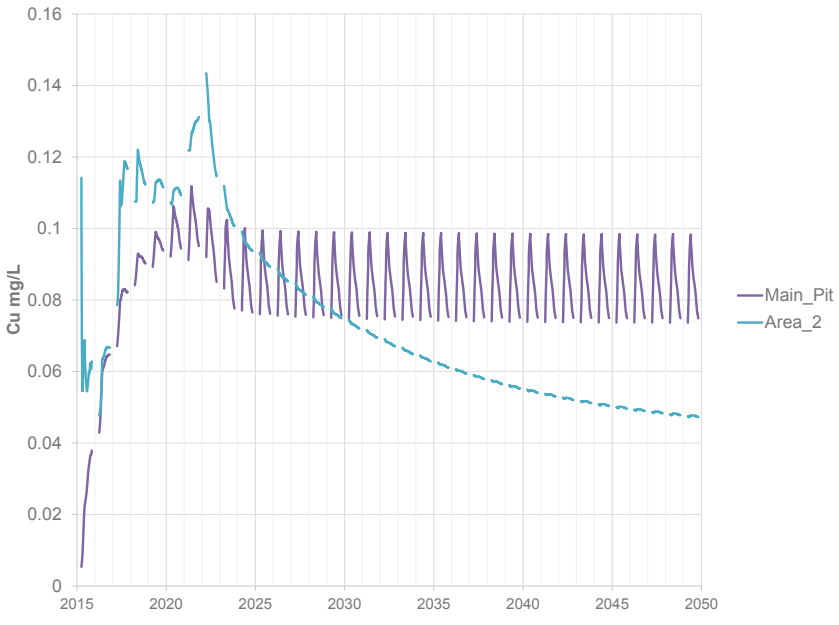

Al mg/L
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1.6
1.4
1.2
1
0.8
0.6
0.4
0.2
0
2015
2020
2025
2030
2035
2040
2045
2050
Main_Pit
Area_2

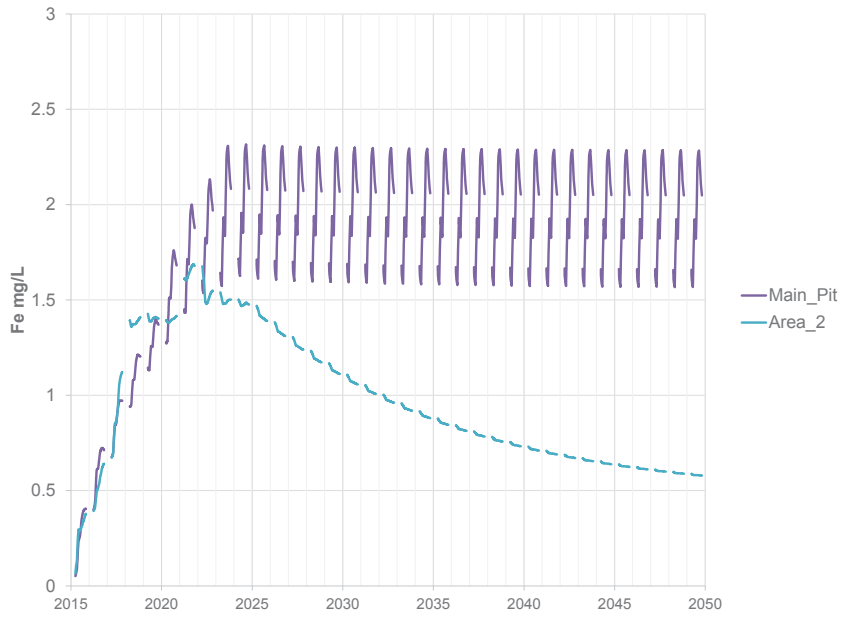

Ammonia mg/L
90
80
70
60
50
40
30
20
10
0
2015
2020
2025
2030
2035
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2045
2050
Main_Pit
Area_2

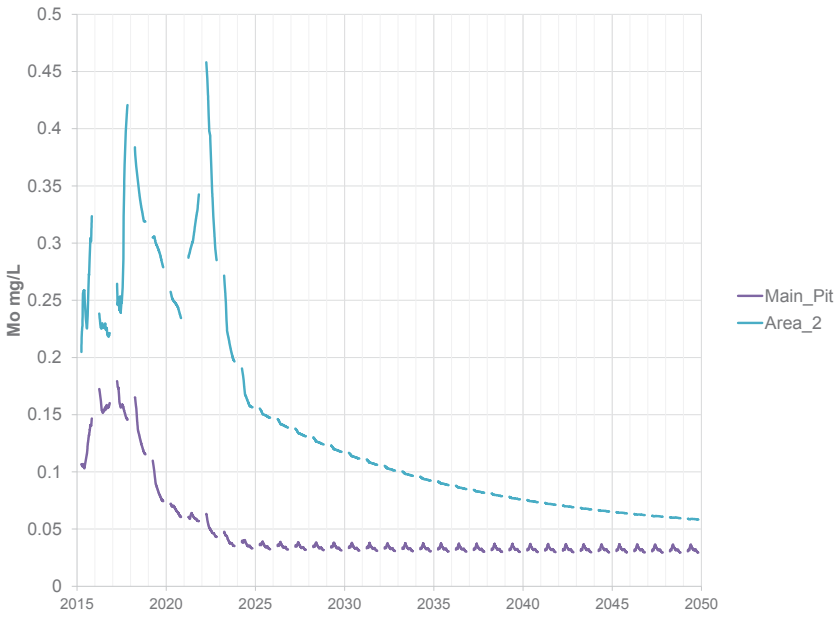

As mg/L
0.014
0.012
0.01
0.008
0.006
0.004
0.002
0
2015
2020
2025
2030
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2040
2045
2050
Main_Pit
Area_2

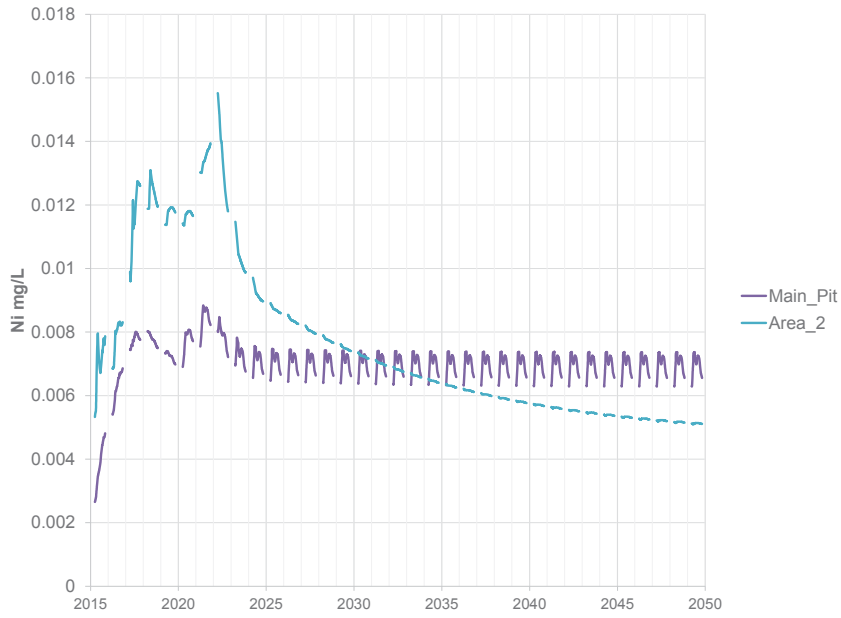

Cd mg/L
7.00E-04
6.00E-04
5.00E-04
4.00E-04
3.00E-04
2.00E-04
1.00E-04
0.00E+00
2015
2020
2025
2030
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2040
2045
2050
Main_Pit
Area_2

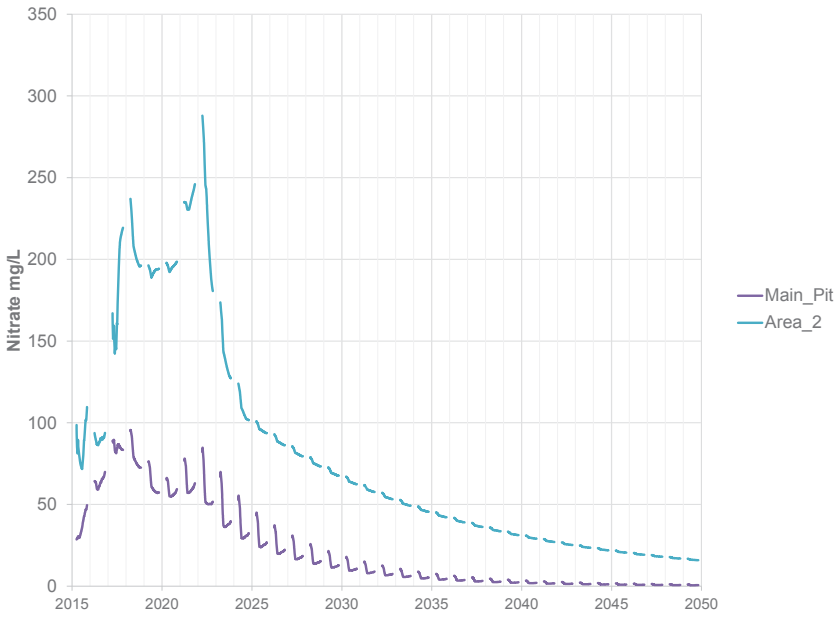

Cr mg/L
0.01
0.009
0.008
0.007
0.006
0.005
0.004
0.003
0.002
0.001
0
2015
2020
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Main_Pit
Area_2

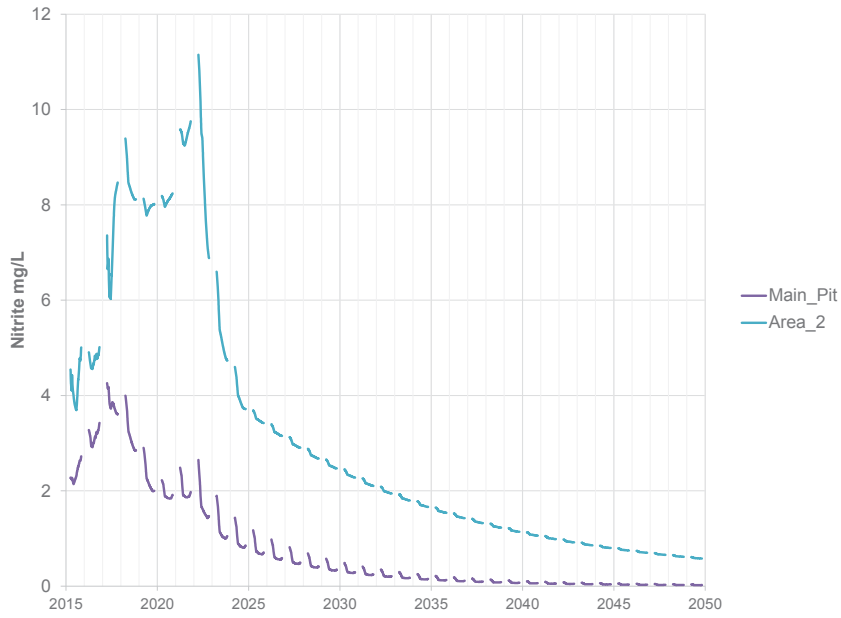

Cu mg/L
0.16
0.14
0.12
0.1
0.08
0.06
0.04
0.02
0
2015
2020
2025
2030
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2040
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Main_Pit
Area_2

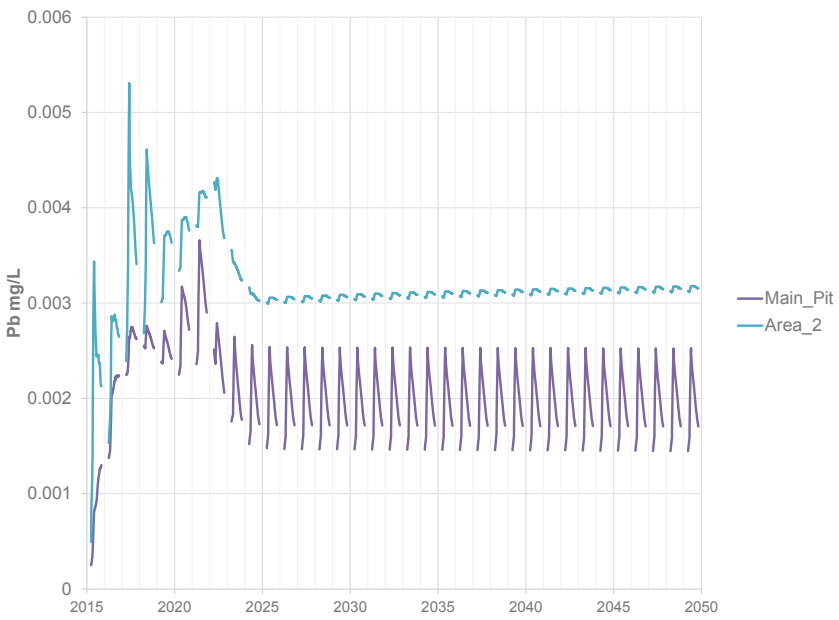

Fe mg/L
3
2.5
2
1.5
1
0.5
0
2015
2020
2025
2030
2035
2040
2045
2050
Main_Pit
Area_2

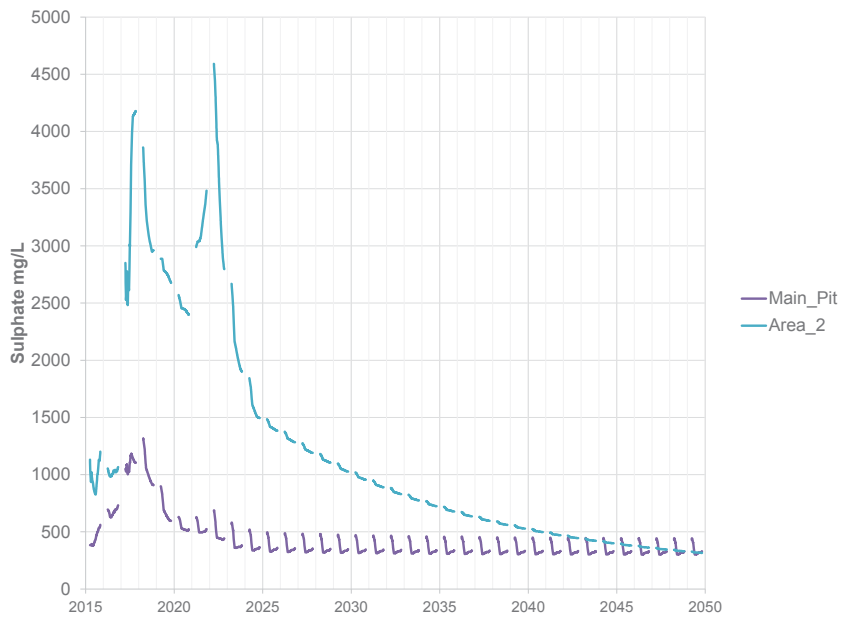


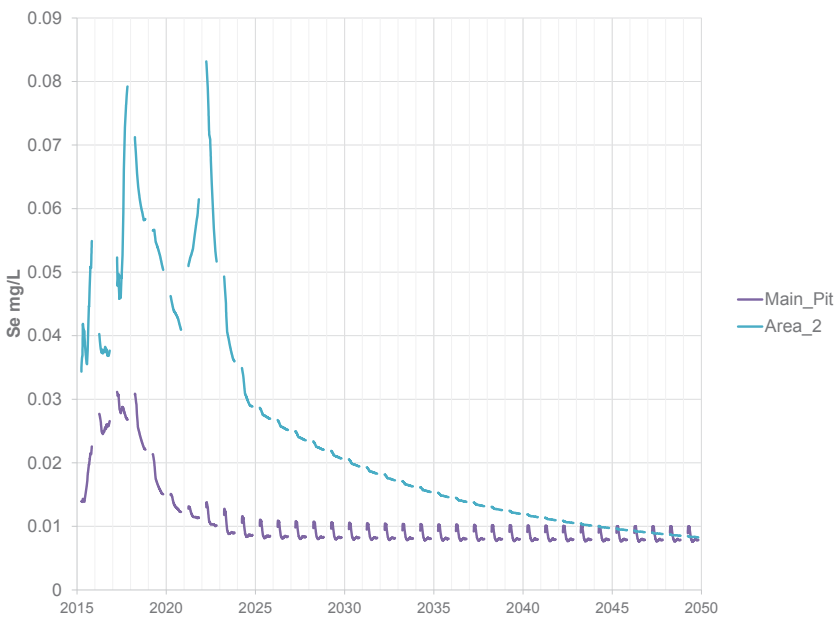


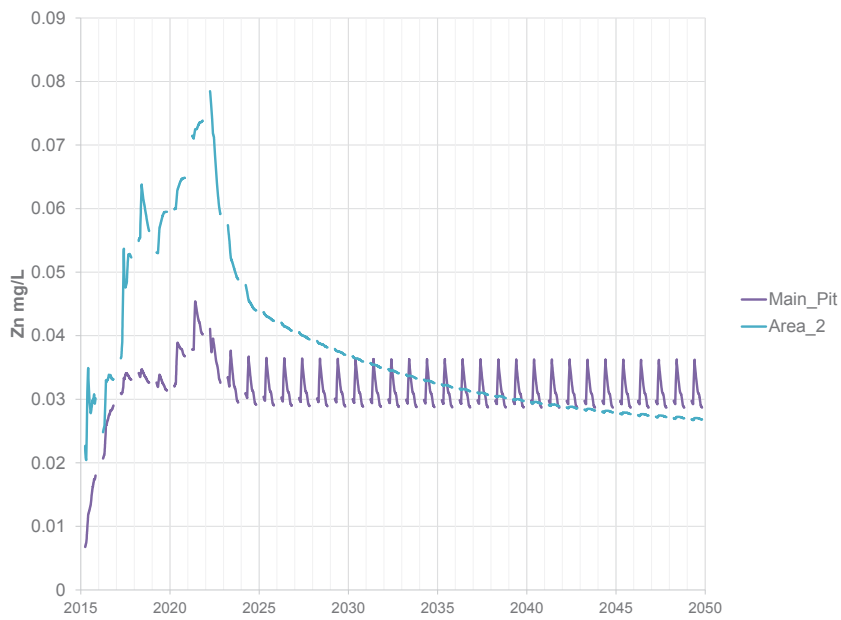




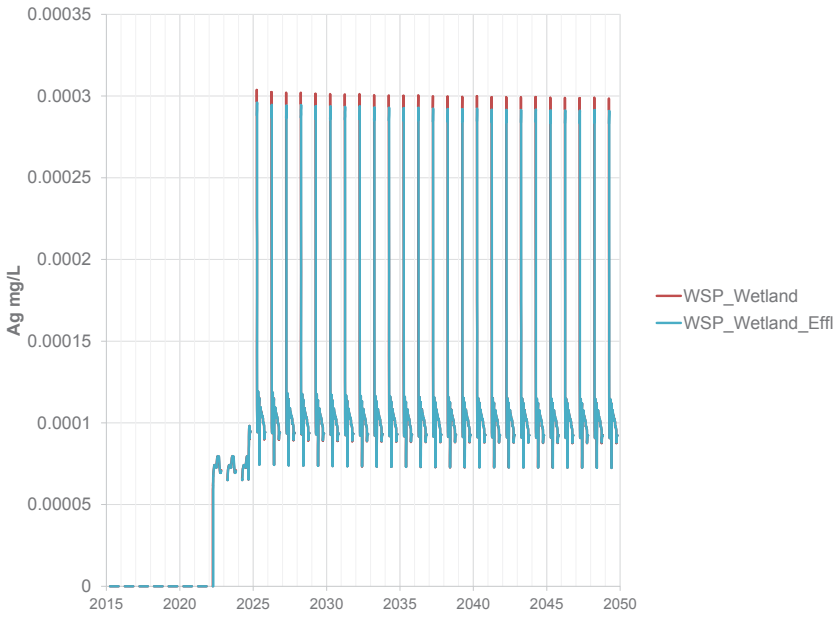


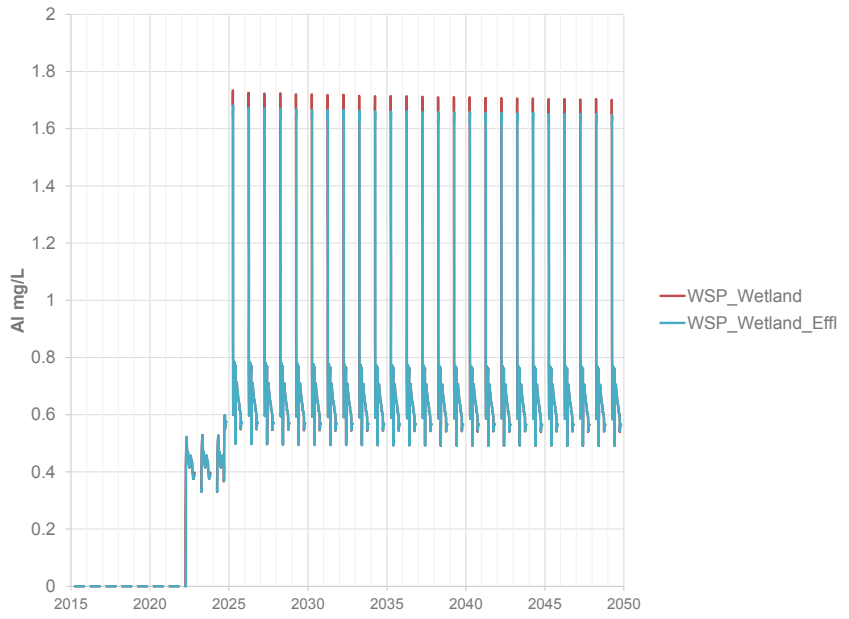


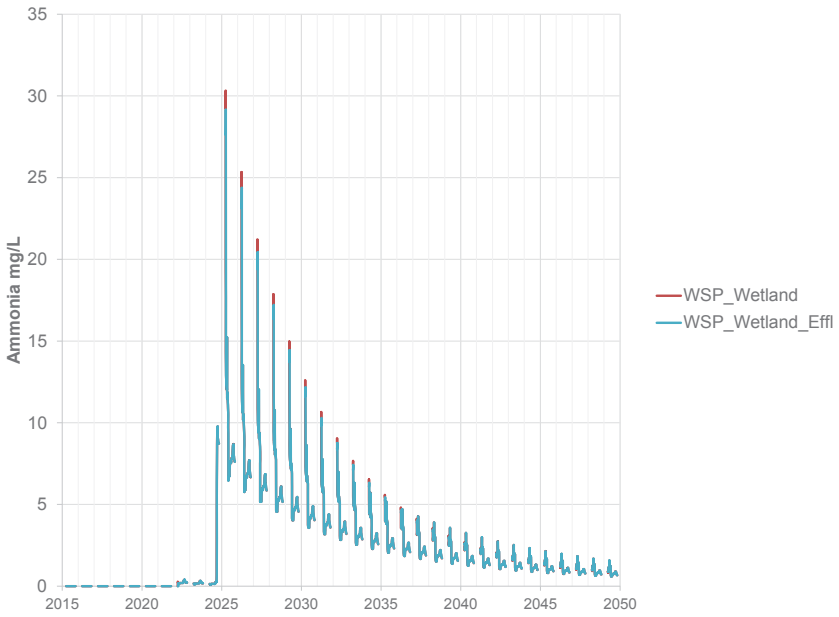


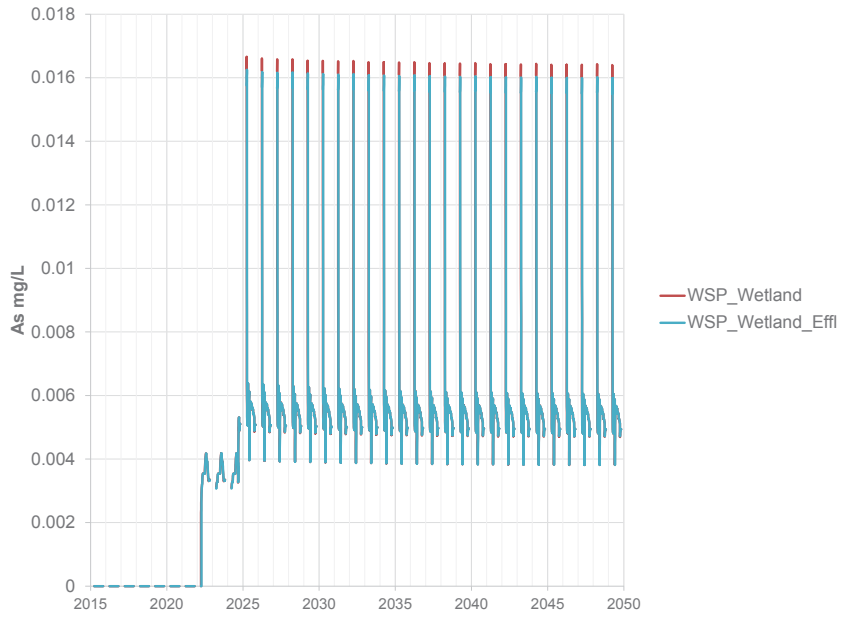


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Model: Minto_2017_1CM002.024_Rev71_SRJ_KNK, 2018/02/06, KNK


Ag mg/L
0.00035
0.0003
0.00025
0.0002
0.00015
0.0001
0.00005
0
2015
2020
2025
2030
2035
2040
2045
2050
WSP_Wetland
WSP_Wetland_Effl

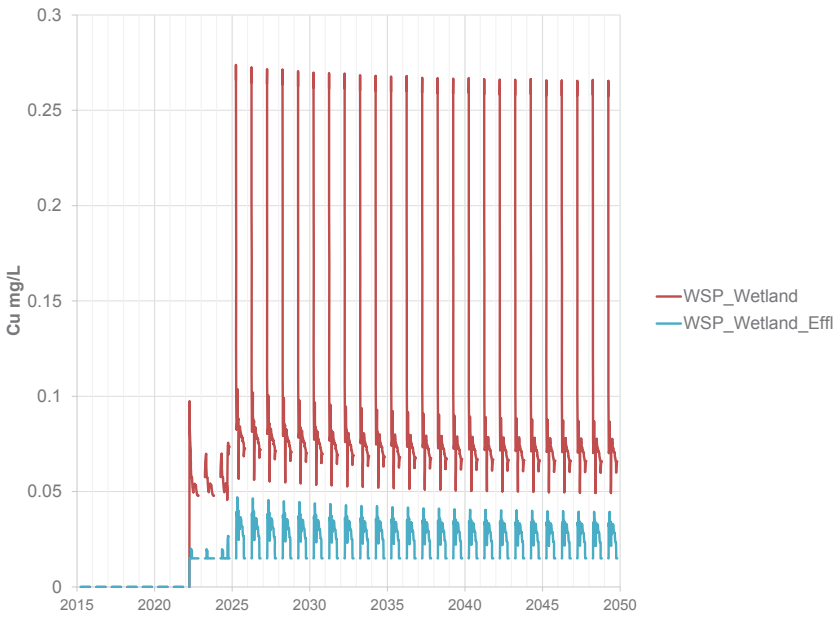

Al mg/L
2
1.8
1.6
1.4
1.2
1
0.8
0.6
0.4
0.2
0
2015
2020
2025
2030
2035
2040
2045
2050
WSP_Wetland
WSP_Wetland_Effl

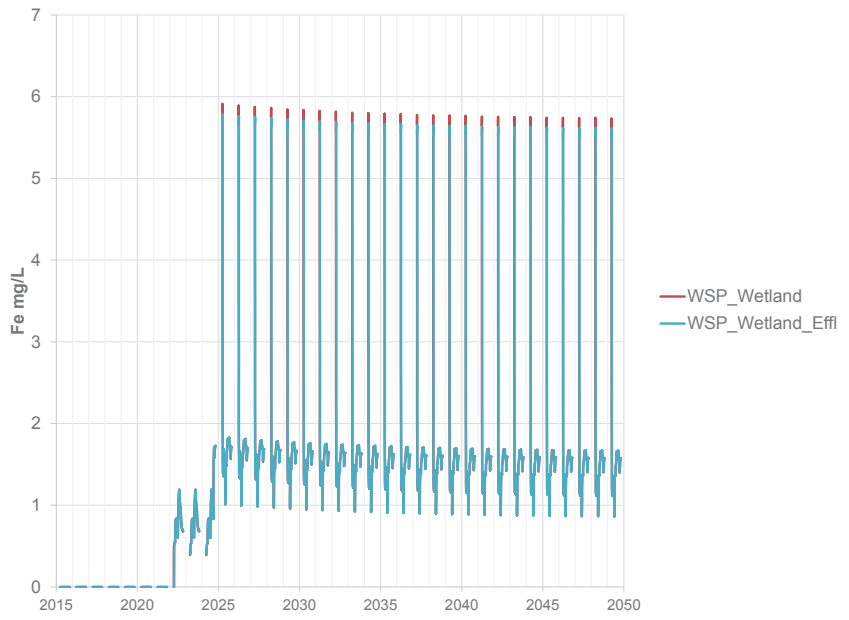

Ammonia mg/L
35
30
25
20
15
10
5
0
2015
2020
2025
2030
2035
2040
2045
2050
WSP_Wetland
WSP_Wetland_Effl

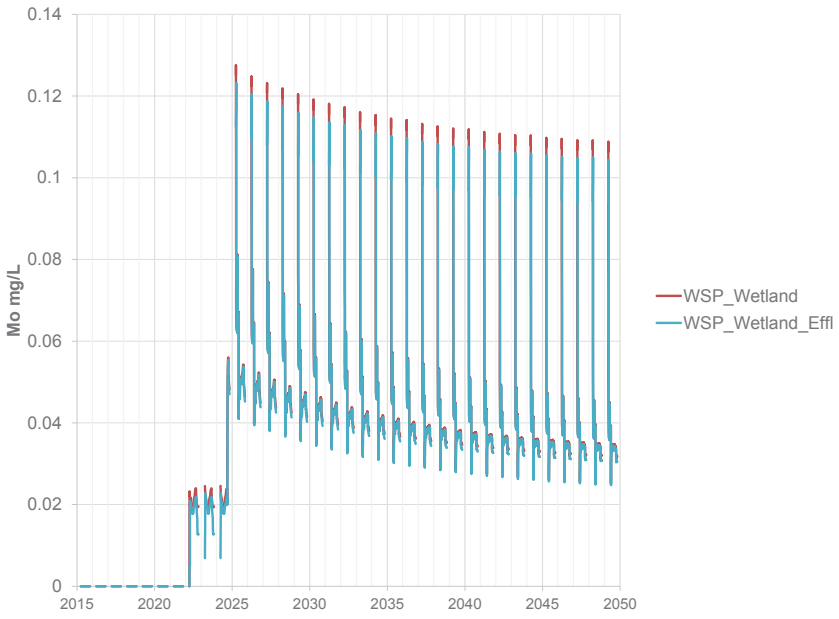

As mg/L
0.018
0.016
0.014
0.012
0.01
0.008
0.006
0.004
0.002
0
2015
2020
2025
2030
2035
2040
2045
2050
WSP_Wetland
WSP_Wetland_Effl

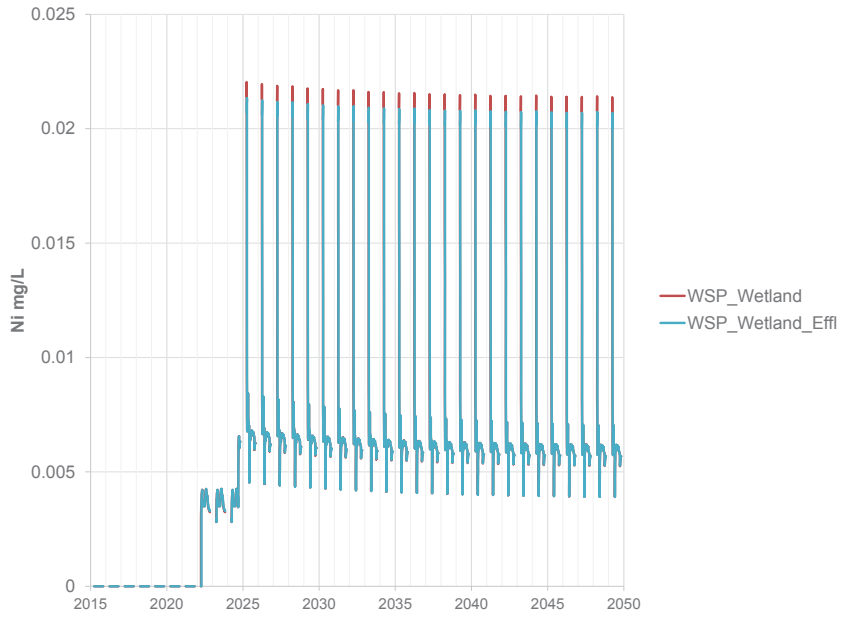
Cd mg/L
0.0012
0.001
0.0008
0.0006
0.0004
0.0002
0
2015
2020
2025
2030
2035
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2050
WSP_Wetland
WSP_Wetland_Effl

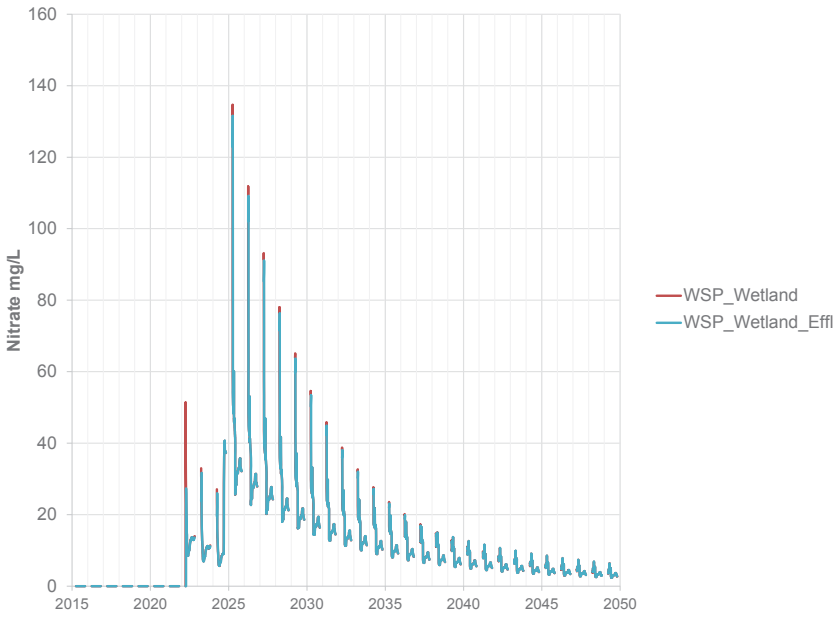
Cr mg/L
0.014
0.012
0.01
0.008
0.006
0.004
0.002
0
2015
2020
2025
2030
2035
2040
2045
2050
WSP_Wetland
WSP_Wetland_Effl

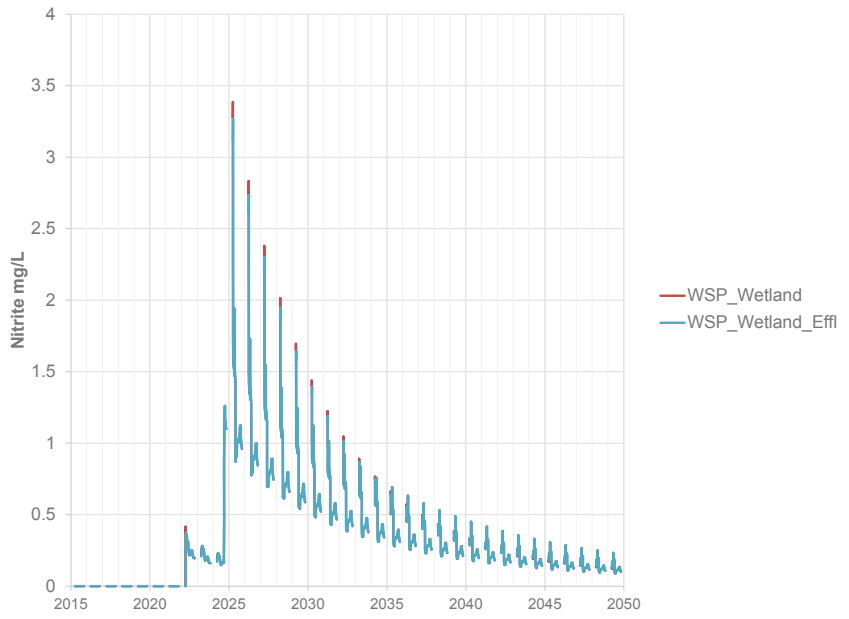

Cu mg/L
0.3
0.25
0.2
0.15
0.1
0.05
0
2015
2020
2025
2030
2035
2040
2045
2050
WSP_Wetland
WSP_Wetland_Effl

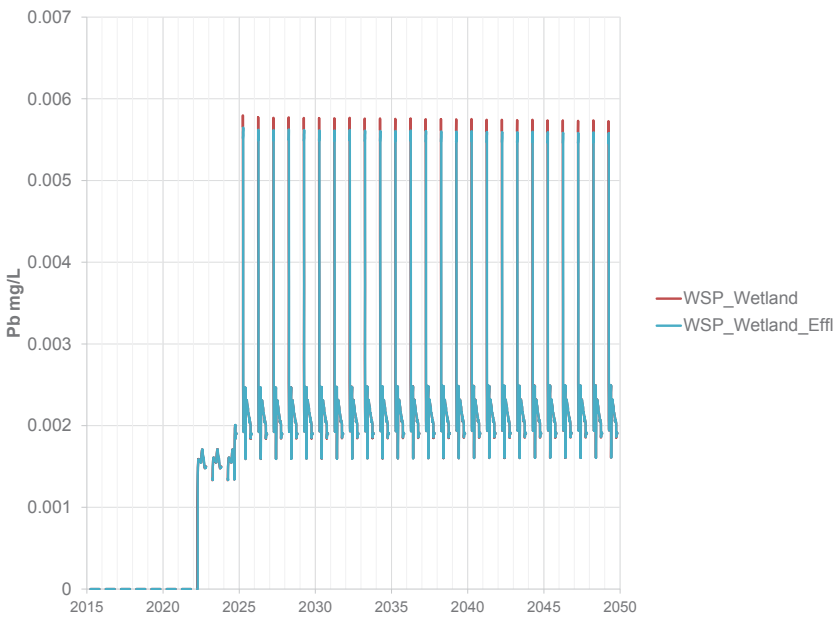

Fe mg/L
7
6
5
4
3
2
1
0
2015
2020
2025
2030
2035
2040
2045
2050
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WSP_Wetland_Effl

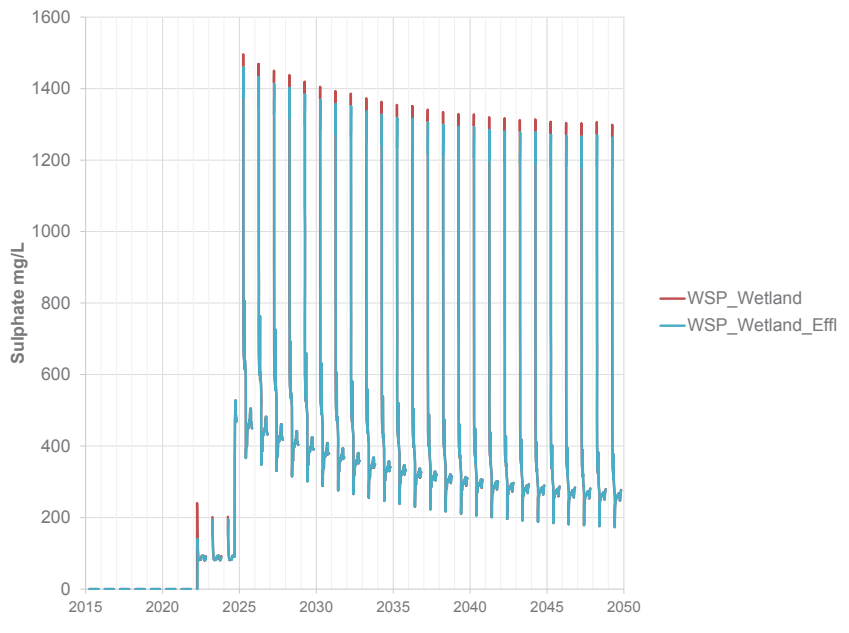


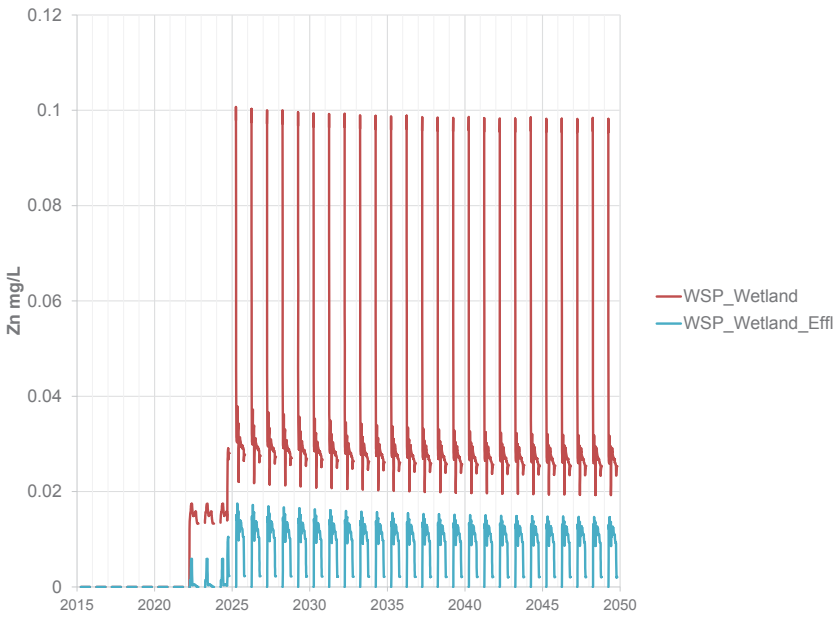


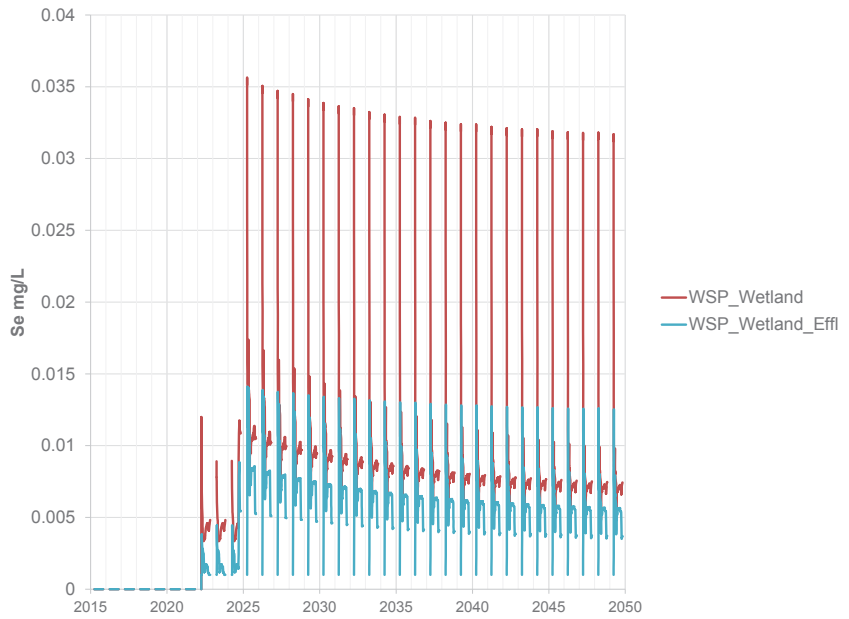




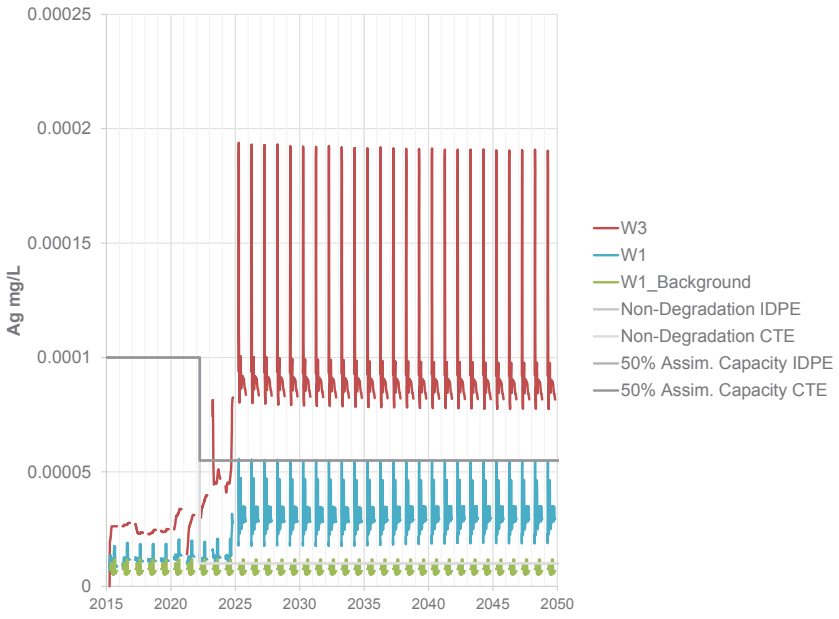


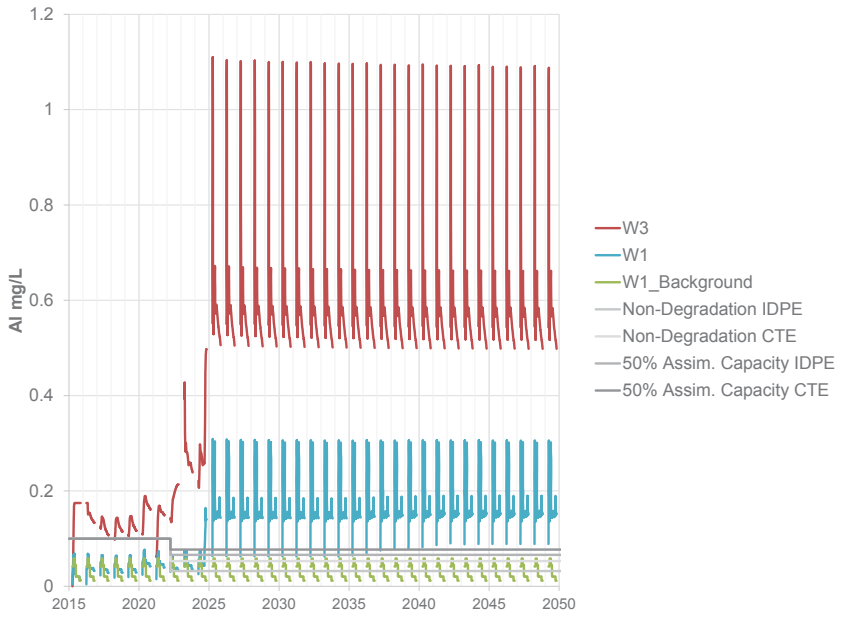


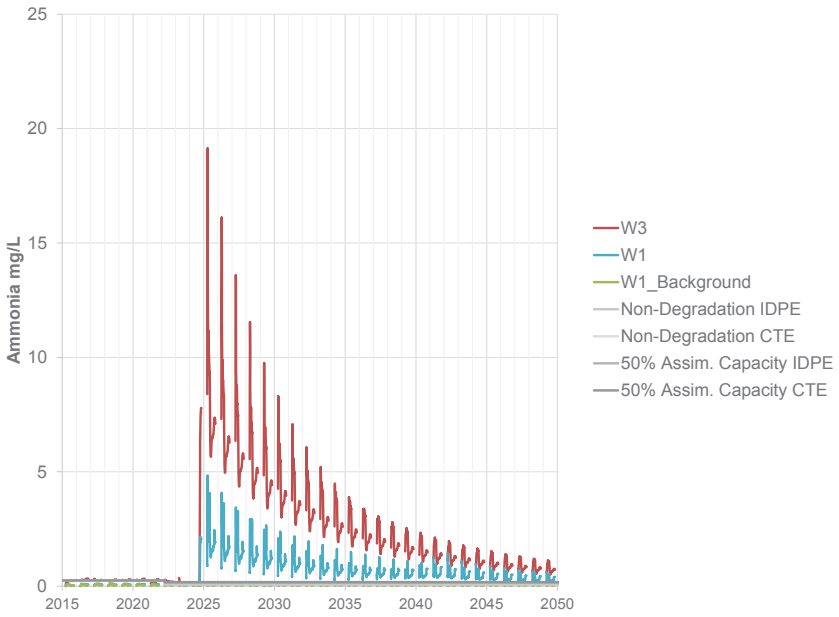


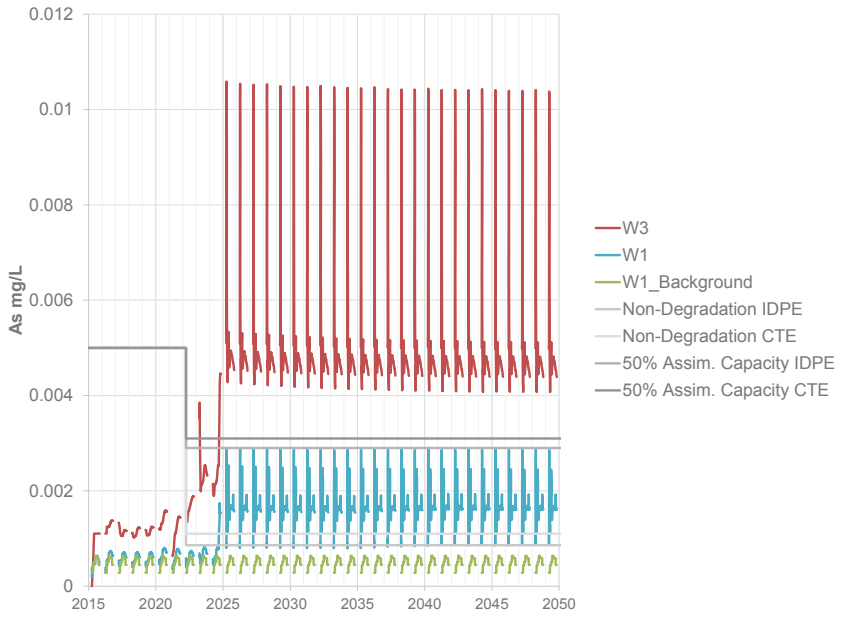


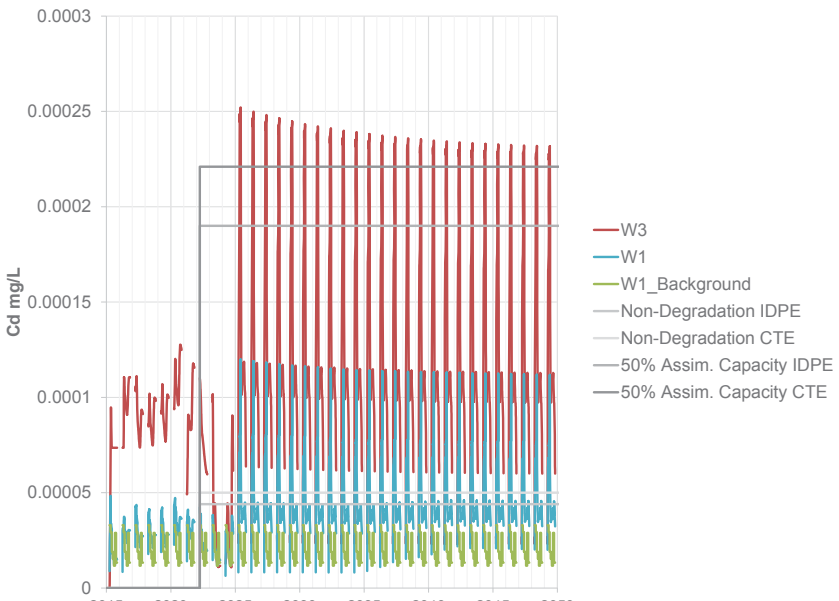
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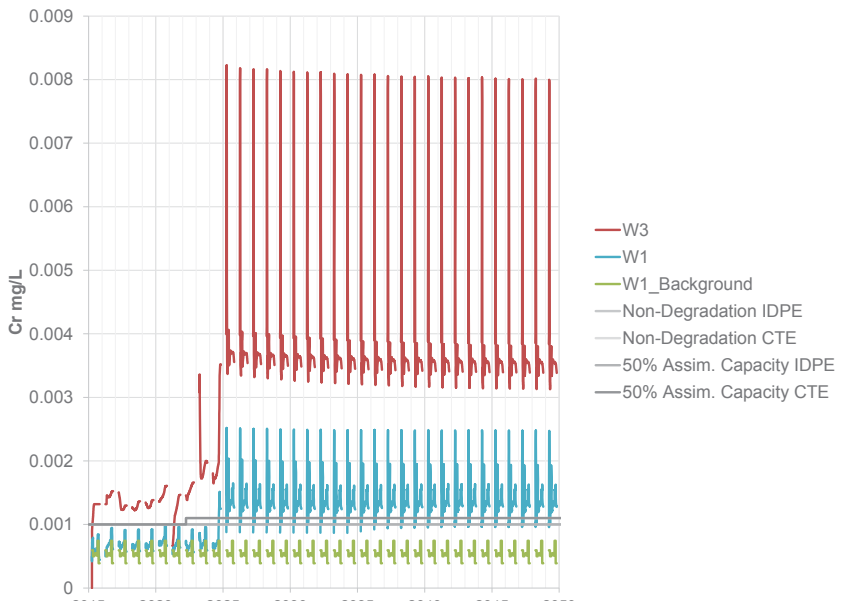


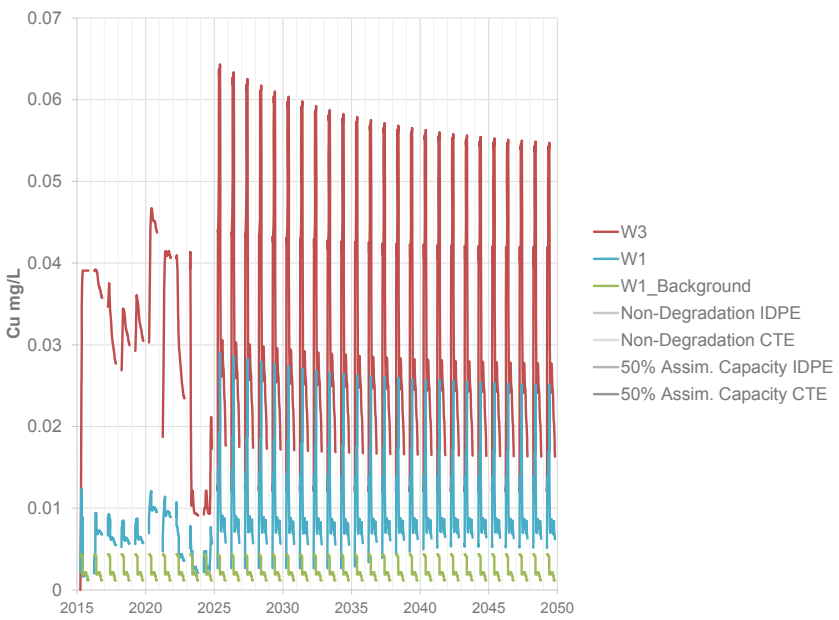


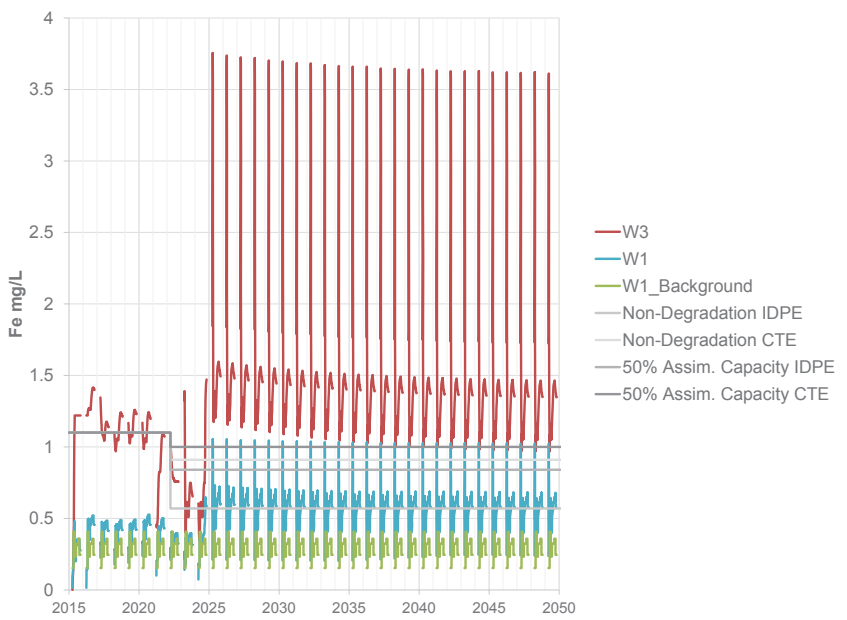




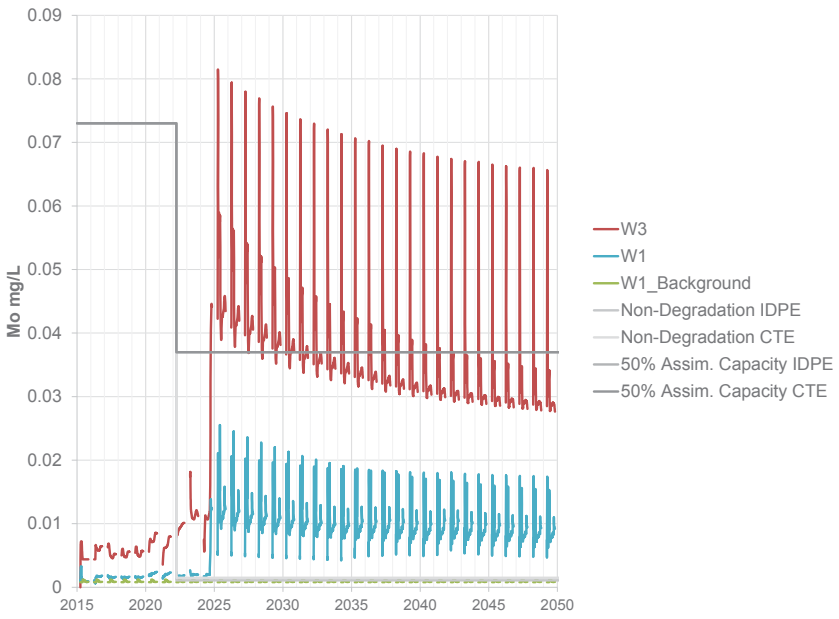


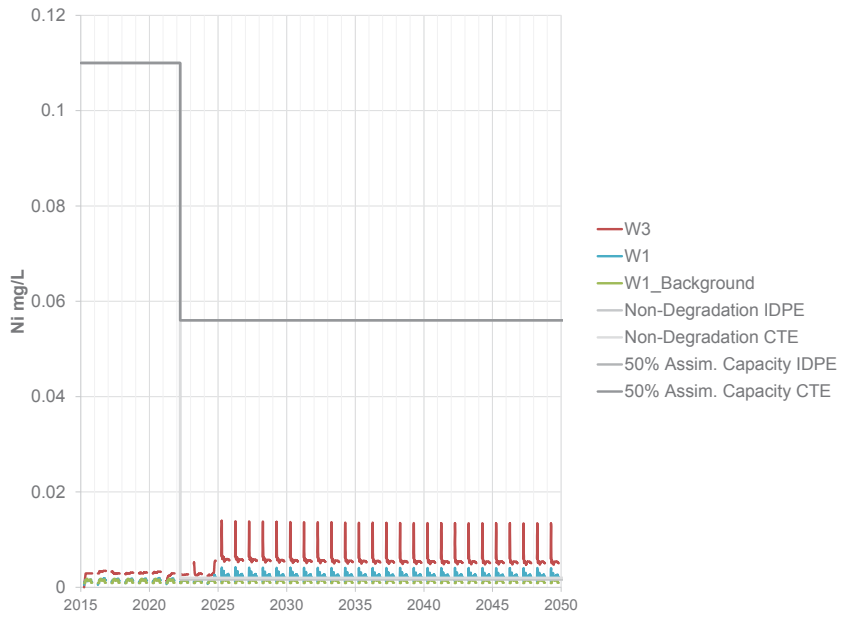


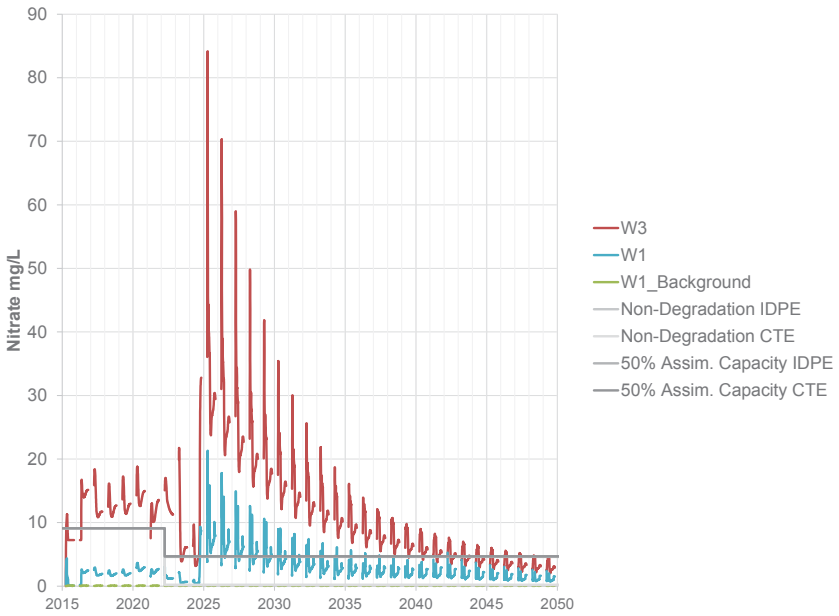


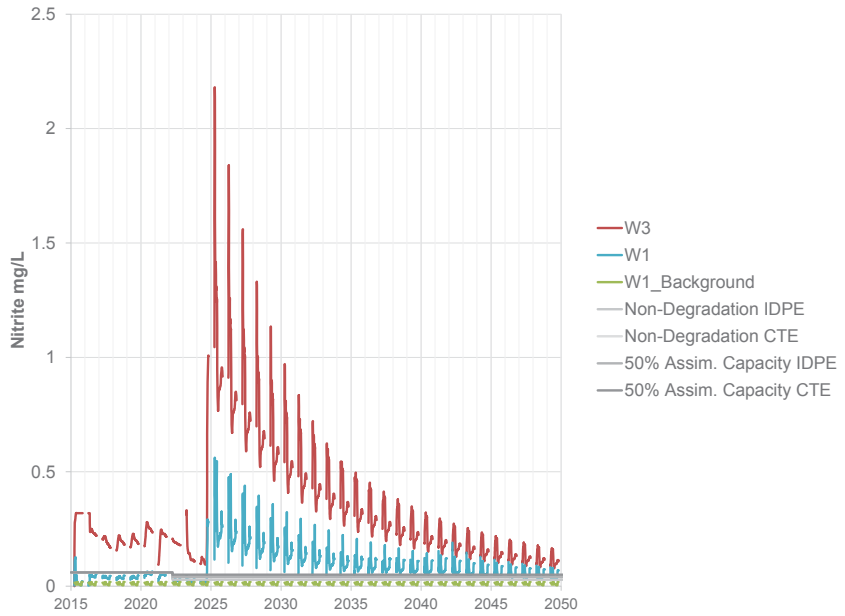


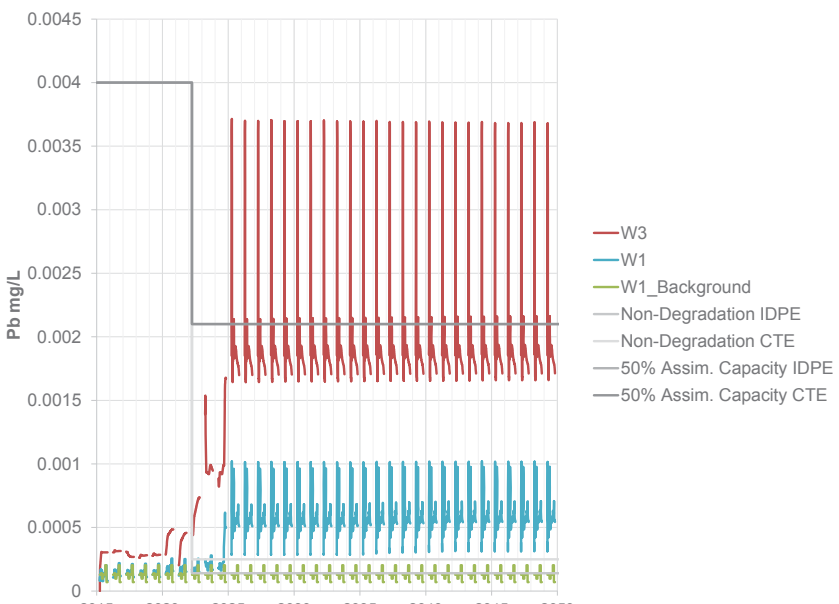
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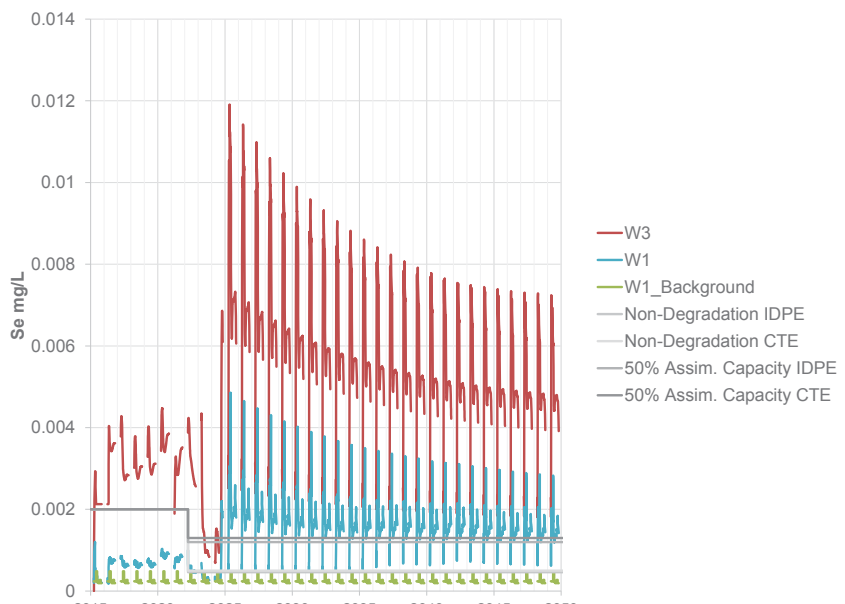


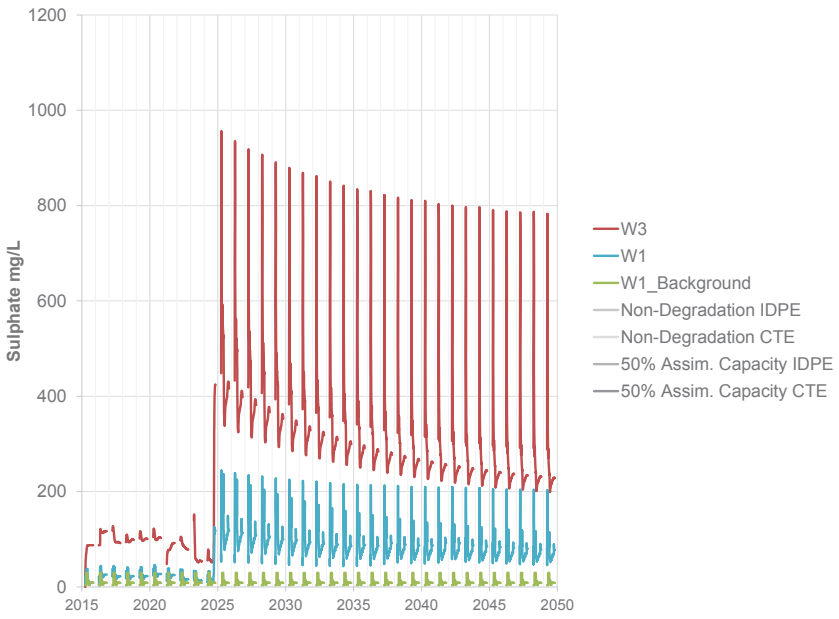


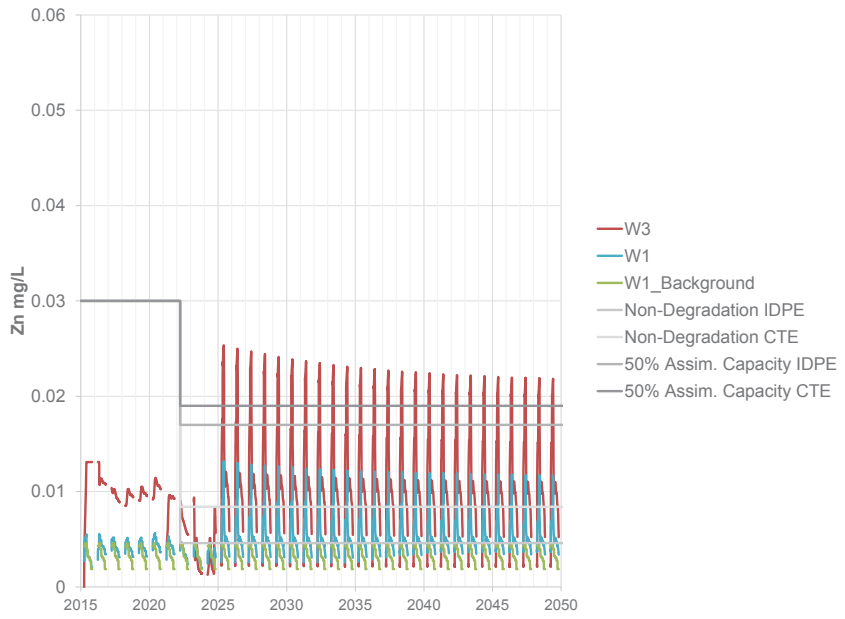












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Model: Minto_2017_1CM002.024_Rev71_SRJ_KNK, 2018/02/06, KNK

Appendix B: Water and Load Balance Source Terms

DSTSF Source Terms

	Expected Case	Worst Case
	mg/L	mg/L
Ag-D	0.00002	0.0001
Al-D	0.01	0.069
Ammonia	0.12	0.98
As-D	0.0005	0.0009
Cd-D	0.00011	0.00024
Cr-D	0.001	0.002
Cu-D	0.094	0.32
Fe-D	0.089	0.39
Mo-D	0.007	0.038
Na-D	24	46
Ni-D	0.0018	0.005
Nitrate	22	84
Nitrite	0.12	2.5
Se-D	0.0039	0.014
Sulphate	120	200
Zn-D	0.005	0.01

POX Ore Tailings Source Term (High Sulphate)

	Expected Case	Worst Case
	mg/kg	mg/kg
Ag-D	0.000016	0.00002
Al-D	0.036	0.097
Ammonia	6.3	20
As-D	0.00049	0.0018
Cd-D	0.000074	0.00016
Cr-D	0.00024	0.00048
Cu-D	0.0052	0.032
Fe-D	0.024	0.049
Mo-D	0.097	0.097
Na-D	250	660
Ni-D	0.0032	0.0032
Nitrate	52	150
Nitrite	3.8	3.8
Se-D	0.032	0.032
Sulphate	2300	4300
Zn-D	0.0024	0.0066

Sulphide Ore Tailings Source Term

	Expected Case	Worst Case
	mg/kg	mg/kg
Ag-D	0.0000041	0.0000082
Al-D	0.016	0.066
Ammonia	1.5	2.8
As-D	0.00017	0.00019
Cd-D	0.0000034	0.0000052
Cr-D	0.000041	0.00041
Cu-D	0.00046	0.00087
Fe-D	0.0041	0.0041
Mo-D	0.04	0.047
Na-D	58	84
Ni-D	0.00061	0.0011
Nitrate	16	22
Nitrite	0.59	1.5
Se-D	0.0074	0.0093
Sulphate	160	300
Zn-D	0.00041	0.002

Waste Rock Volume-based Source Terms

	Expected Case	Worst Case
	mg/m3/yr	mg/m3/yr
Ag-D	0.0005	0.0005
Al-D	0.55	4
Ammonia	1.7	5.7
As-D	0.012	0.025
Cd-D	0.00035	0.0018
Cr-D	0.025	0.025
Cu-D	0.45	1.2
Fe-D	7	26
Mo-D	0.067	0.16
Na-D	250	550
Ni-D	0.025	0.064
Nitrate	150	940
Nitrite	1.1	7.5
Se-D	0.034	0.22
Sulphate	1400	4800
Zn-D	0.12	0.22

Note: load distributed over the year based on the mean hydrograph

Waste Rock Area-based Source Terms

	Expected Case	Worst Case
	mg/L	mg/L
Ag-D	0.000062	0.000062
Al-D	0.068	0.49
Ammonia	0.22	0.71
As-D	0.0015	0.0031
Cd-D	0.000043	0.00023
Cr-D	0.0031	0.0031
Cu-D	0.055	0.15
Fe-D	0.87	3.2
Mo-D	0.0083	0.02
Na-D	31	68
Ni-D	0.0031	0.0079
Nitrate	19	120
Nitrite	0.14	0.92
Se-D	0.0042	0.027
Sulphate	170	600
Zn-D	0.015	0.028

Pit Walls

	Expected Case	Worst Case
	mg/yr	mg/yr
Ag-D	7,700	20,000
Al-D	73,000,000	140,000,000
Ammonia	0	0
As-D	390,000	1,100,000
Cd-D	9,900	49,000
Cr-D	160,000	700,000
Cu-D	2,300,000	6,000,000
Fe-D	8,400,000	25,000,000
Mo-D	1,400,000	5,800,000
Na-D	0	0
Ni-D	120,000	550,000
Nitrate	0	0
Nitrite	0	0
Se-D	290,000	550,000
Sulphate	3,300,000,000	6,600,000,000
Zn-D	740,000	3,500,000

Note: load distributed over the year based on the mean hydrograph

Underground East Source Terms

	Expected Case	Worst Case
	mg/L	mg/L
Ag-D	0.00002	0.00002
Al-D	0.0045	0.016
Ammonia	0.044	0.21
As-D	0.0005	0.0016
Cd-D	0.00001	0.000024
Cr-D	0.00097	0.001
Cu-D	0.0002	0.0031
Fe-D	0.04	3
Mo-D	0.001	0.007
Na-D	110	150
Ni-D	0.001	0.0011
Nitrate	0.023	0.052
Nitrite	0.053	0.27
Se-D	0.0001	0.00098
Sulphate	800	1200
Zn-D	0.005	0.027

Underground South Source Terms

	Expected Case	Worst Case
	mg/L	mg/L
Ag-D	0.00002	0.000031
Al-D	0.011	0.024
Ammonia	18	91
As-D	0.0016	0.0043
Cd-D	0.000017	0.00028
Cr-D	0.001	0.001
Cu-D	0.0074	0.067
Fe-D	0.012	0.076
Mo-D	0.018	0.042
Na-D	36	84
Ni-D	0.0023	0.0082
Nitrate	35	180
Nitrite	1.4	7.7
Se-D	0.00048	0.0044
Sulphate	370	1200
Zn-D	0.0061	0.058

Upper and Lower Minto Creek Background Water Quality

Month	Ag-D	Al-D	Ammonia	As-D	Cd-D	Cr-D	Cu-D	Fe-D	Mn-D	Mo-D	Ni-D	Nitrate	Nitrite	Se-D	Sulphate	Zn-D
	mg/L	mg/L	mg/L	mg/L	mg/L	mg/L	mg/L	mg/L	mg/L	mg/L	Hg	mg/L	mg/L	mg/L	mg/L	mg/L
January	0.00001	0.0053	0.005	0.00041	0.000035	0.0005	0.0011	0.094	0.034	0.0015	0.0005	0.32	0.0025	0.00043	24	0.0025
February	0.00001	0.021	0.006	0.00033	0.000064	0.0005	0.015	0.049	0.0071	0.0014	0.0005	0.29	0.0025	0.00034	12	0.0083
March	0.00001	0.0031	0.0067	0.00039	0.000005	0.0005	0.0012	0.012	0.0042	0.0011	0.0005	0.2	0.0025	0.00031	23	0.0025
April	0.0000087	0.038	0.012	0.00028	0.000031	0.00059	0.0041	0.15	0.041	0.00078	0.00084	0.035	0.015	0.00021	3.7	0.004
May	0.00001	0.05	0.06	0.00042	0.000024	0.0005	0.0039	0.36	0.095	0.00061	0.0013	0.031	0.0036	0.00018	29	0.0043
June	0.0000071	0.026	0.062	0.00047	0.000018	0.00055	0.0017	0.2	0.055	0.0011	0.0013	0.1	0.015	0.00037	14	0.0038
July	0.0000088	0.02	0.048	0.00061	0.000015	0.00059	0.002	0.32	0.068	0.001	0.0013	0.12	0.0091	0.00021	4.4	0.0032
August	0.000018	0.02	0.033	0.00063	0.000012	0.00063	0.0021	0.36	0.085	0.00096	0.0014	0.11	0.008	0.00021	9.2	0.0031
September	0.000009	0.02	0.024	0.00058	0.000016	0.00084	0.0017	0.36	0.079	0.00083	0.0016	0.046	0.0038	0.00027	9.5	0.0027
October	0.0000086	0.013	0.034	0.0005	0.000015	0.00044	0.0013	0.28	0.089	0.00095	0.0011	0.087	0.021	0.0002	8.6	0.002
November	0.00001	0.01	0.051	0.001	0.000011	0.00059	0.0013	1.7	1.1	0.0011	0.0017	0.12	0.005	0.00016	8.4	0.0025
December	0.00001	0.005	0.021	0.00054	0.0000065	0.0005	0.0017	0.046	0.068	0.001	0.00095	0.086	0.0025	0.0003	36	0.0025

Minto North Background Water Quality

Month	Ag-D	Al-D	Ammonia	As-D	Cd-D	Cr-D	Cu-D	Fe-D	Mn-D	Mo-D	Ni-D	Nitrate	Nitrite	Se-D	Sulphate	Zn-D
	mg/L	mg/L	mg/L	mg/L	mg/L	mg/L	mg/L	mg/L	mg/L	mg/L	Hg	mg/L	mg/L	mg/L	mg/L	mg/L
January	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0
February	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0
March	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0
April	0.000005	0.041	0.038	0.00038	0.000027	0.00022	0.0026	0.13	0.0073	0.00038	0.00099	0.02	0.005	0.00009	0.5	0.0034
May	0.000005	0.028	0.0065	0.00038	0.0000085	0.0003	0.0024	0.16	0.0093	0.00046	0.0012	0.02	0.005	0.00013	0.5	0.0025
June	0.000005	0.014	0.019	0.00038	0.000006	0.0003	0.0019	0.097	0.0035	0.00082	0.001	0.06	0.005	0.00015	5.9	0.0018
July	0.000005	0.018	0.022	0.00042	0.00001	0.00033	0.0019	0.082	0.0017	0.00083	0.001	0.048	0.005	0.00015	3.2	0.0012
August	0.000005	0.022	0.02	0.00056	0.000007	0.0004	0.0019	0.2	0.0021	0.00086	0.0013	0.062	0.005	0.00013	2.8	0.00096
September	0.00002	0.019	0.02	0.00043	0.0000065	0.00035	0.0019	0.17	0.0067	0.00084	0.0012	0.042	0.005	0.00013	5.4	0.0011
October	0.000005	0.016	0.0093	0.00039	0.00001	0.0002	0.0016	0.11	0.0064	0.00085	0.00098	0.09	0.005	0.00017	6.8	0.0007
November	0.0000025	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0
December	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0

Appendix C: Minto Mine 2016 Water Balance Update

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Memo

To:	Deborah Flemming, Ryan Herbert	**Client:**	Minto Explorations Ltd.
From:	Soren Jensen, Kaitlyn Kooy	**Project No:**	1CM002.024
Cc:	Dylan MacGregor (SRK)	**Date:**	March 28, 2017
Subject:	2016 Water Balance and Water Quality Model Summary for the Minto Mine Site		

1 Introduction and Background

This memorandum provides a summary of the 2016 water balance and water quality model updates for the Minto Mine site. The update covers the period January 1, 2016 through December 31, 2016.

The water balance update includes a review and summary of precipitation, flow and water inventory data for the Mine site. The water quality update includes a comparison of water quality data collected in 2016 to updated water quality model predictions for Phase V/VI of the Mine development. Updated water quality predictions for the Main Pit Tailings Management Facility (MPTMF) and the Water Storage Pond (WSP) are provided for the 2018 and for the post-closure period when predicted concentrations are the same from year to year (steady state concentrations).

2 Water Balance Update

2.1 Precipitation

Table 1 shows a summary of monthly precipitation measured at the Mine site between October 2015 and December 2016 along with precipitation data from the regional station at Carmacks, YT (Climate ID: 2100301)[1], which is located 77 km Southwest of the Minto Mine. In the past, the meteorological station at Pelly Ranch (Climate ID 2100880)[1] located 25 km north of Minto was used as a regional reference station. However, the published data record from Pelly Ranch ends in March 2015 so the Carmacks Station was used instead.

Approximately 242 mm of precipitation was collected at the Mine site in the 2016 hydrological year, which was nearly identical to the annual precipitation of 243 mm measured in 2015. A 242 mm annual total precipitation roughly corresponds to a 1 in 15 dry year.

[1] Pelly Ranch Data: obtained from Meteorological Service of Canada, Environment Canada.

Table 1: Precipitation Records for the Minto Mine Site and Pelly Ranch

		Campbell Scientific Station (Minto Mine)		
Year	**Month**	**Tipping Bucket Gauge**	**Geonor Gauge[A]**	**Carmacks Ranch[B] (Climate ID 2100301)**
		mm/month	mm/month	mm/month
2015	Oct	14.7	27.7	n/a
2015	Nov	14.3	7.1	n/a
2015	Dec	9.5	11.5	n/a
2016	Jan	9.1	11.7	7.5
2016	Feb	6.9	7.8	12.2
2016	Mar	2.6	3.3	5.0
2016	Apr	14.7	14.7	13.1
2016	May	39.3	31.5	28.5
2016	Jun	30.2	22.2	17.1
2016	Jul	74.7	68.2	94.9
2016	Aug	46.0	35.9	26.1
2016	Sept	18.5	17.8	9.6
2016	Oct	6.7	9.8	8.4
2016	Nov	24.2	13.6	14.4
2016	Dec	7.4	14.1	17.5
SUM Hydrological Year, Nov. 2015 to Oct. 2016		273	242	256

Source: Minto Site Data: X:\01_SITES\Minto\1CM002.024_Water_Balance_Support\2015_Water_Balance_Update\Minto Water Balance\2016 Met Station Data Summary.xlsx

Notes:
A: Tipping bucket measurement used for month of November.
B: Data obtained from Meteorological Service of Canada, Environment Canada.
n/a: Not available at time of publication.

Minto's Campbell Scientific meteorological station measures total precipitation using a Geonor and a tipping bucket rain gauge. From October through May, the tipping bucket is equipped with a snowfall conversion adaptor, which allows it to measure snowfall as snow water equivalent. The Geonor precipitation gauge collects precipitation in a bucket and records precipitation by measuring the weight of the bucket. In the winter months, the bucket is partially filled with an antifreeze solution that melts any snow collected. Figure 1 shows monthly precipitation recorded by the two gauges as well as total precipitation recorded at Carmacks.

The total annual precipitation measured by the two gauges at Minto differ by 12%. Both gauges at Minto are within 6% of the total precipitation measured at Carmacks on an annual basis. The precipitation measured by the Geonor gauge at Minto was considered more reliable than the tipping bucket gauge and was therefore used for the hydrological analysis.

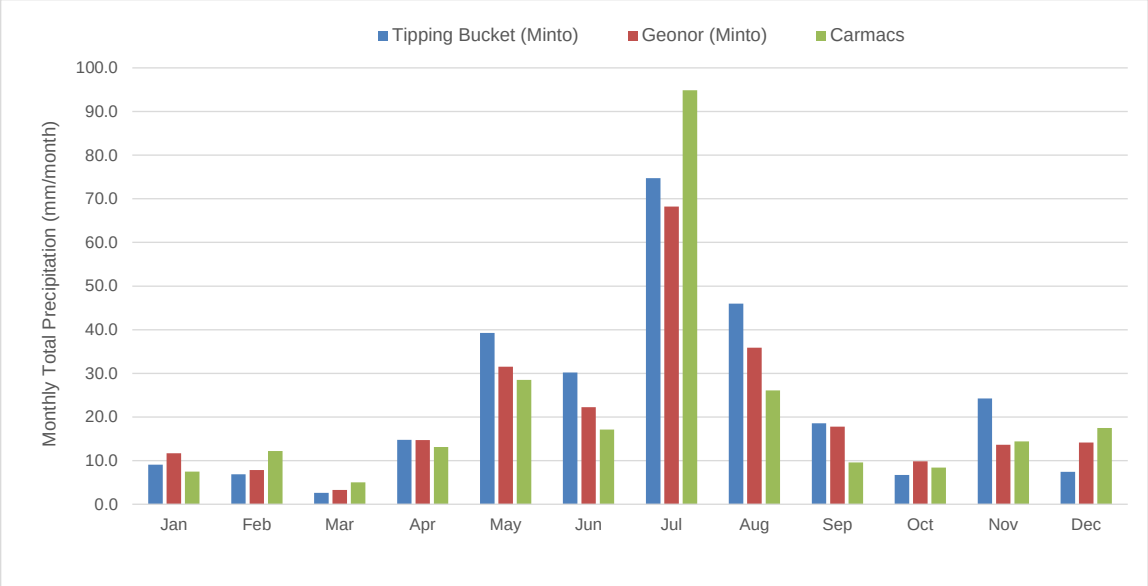


Figure 1: 2015 Monthly Total Precipitation Measurements at Minto and at Carmacks in 2016

2.2 Snow Course Data

Snow course surveys were completed at the three established snow survey stations at the Mine site in 2016. Table 2 shows a summary of the snow survey data (i.e. an average of the results from the three stations) from 2009 to 2016. The depth and water equivalent of the snow pack provides an indication of the volume of surface runoff that must be managed the following freshet. In April and May 2016, approximately 260,000 m^3 of surface runoff was collected from catchments at the Mine site upstream of the Water Storage Dam. This volume corresponds to roughly 25 mm of runoff, which was slightly greater than the estimated 22 mm of runoff observed in 2015.

Table 2: Summary of Snow Survey Data for the Minto Mine Site

Year	February			March			April		
	Snow Depth (cm)	Snow Density (%)	Water Equivalent (mm)	Snow Depth (cm)	Snow Density (%)	Water Equivalent (mm)	Snow Depth (cm)	Snow Density (%)	Water Equivalent (mm)
2009	55.6	16.6	92.7	70.2	15.7	110.0	67.4	22.3	150.7
2010	60.5	17.8	107.7	58.1	20.7	120.7	40.4	[A]13.9	56.0
2011	57.2	18.7	106.0	70.3	20.1	141.7	52.3	22.8	111.7
2012	54.7	20.3	111.0	64.6	19.6	127.0	61.3	21.5	132.7
2013	58.7	15.7	91.3	45.8	25.0	106.0	33.7	15.4	62.7
2014	44.3	19.0	84.3	45.8	22.3	99.7	41.0	25.7	67.3
2015	44.3	20.7	90.3	25.3	29.0	76.6	30.0	23	67.8
2016	40.7	19.0	80.3	42.3	23.0	80.7	15.5	n/a	n/a

Source: X:\01_SITES\Minto\1CM002.024_Water_Balance_Support\2016_Water_Balance_Update\MintoSnowMaster_2016.xlsx

Notes:

n/a – not available.
A: zero snow at #3, density is an average of snowpack at #1 and #2, average depth and water-equivalent is average of all three sites.

2.3 Water Management

Water that is suitable for release into Minto Creek is conveyed to the Water Storage Pond (WSP), while water collected from active mine areas is routed to the Main Pit Tailings Management Facility (MPTMF) or the Area 2 Pit Tailings Management Facility (A2PTMF). Since November 2012, the MPTMF has also been used for subaqueous deposition of tailings. Deposition of mine water and tailings (subaqueous) to the A2PTMF commenced in April 2015.

Other water management features on the Mine site include:

- W15 sump: collects surface runoff and seepage from:
  - The Southwest Waste Dump;
  - Part of the Main Waste Dump; and
  - Adjacent undisturbed catchments.

  In 2016, water collected at W15 was routed to the Main Pit TMF in early spring (March and most of April). Between April 27 and October 3, water from W15 was routed to the WSP. Approximately 50% of the catchment that report to W15 is covered by waste rock. Therefore, it is notable that water quality parameter concentrations are low enough that the water is suitable for discharge to Quarry Creek.

- W35a sump: collects surface runoff from the minimally disturbed southern catchments. Water collected at W35a in 2016 was piped to the Mai Pit in early April. From April 14 to October 3 the water was conveyed to the WSP. The remainder of the year the W35 water was routed to the Main Pit.

- W62 sump (formerly known as W36 or W37 sump): collects surface runoff and seepage from the mill valley, including contributions from the Dry Stack Tailings Storage Facility. Water collected at the W62 sump is pumped to the MPTMF.
- South Diversion Ditch: diverts water from minimally disturbed southern catchments to the WSP (can also be routed to the MPTMF).
- WSP: reservoir for water that meets discharge criteria and is destined for discharge to Minto Creek.

2.4 2016 Water Balance

Table 3 summarizes the monthly water and tailings inventory in Minto's MPTMF and A2PTMF as well as water inventory in the WSP. In 2016, the water inventory in the MPTMF was reduced by approximately 180,000 m^3, while the water inventory in A2PTMF increased by roughly 530,000 m^3. The increase in water inventory in the A2PTMF roughly corresponds to the volume of pore water in the sub-aqueously deposited tailings in 2016. The WSP water inventory was reduced by about 15,000 m^3 between January 1 and December 31, 2016.

Table 4 shows a summary of the 2016 water balance for the Mine site. The total surface runoff collected on site was estimated to be 590,000 m^3 based on the change in the water inventory and the known volume of water released to Minto Creek. Including an estimated inflow of 30,000 m^3 of groundwater, the total site-wide yield was estimated at about 620,000 m^3 for the year. The total catchment upstream of the Water Storage Dam measures approximately 1,040 ha. Therefore, 620,000 m^3 of runoff from 1,040 ha gives a unit yield of approximately 60 mm/year.

The water and load balance model used for forecasting surface runoff volumes uses a site-wide annual average runoff coefficient, which has been derived based on previous years' water balance results. The runoff coefficient is estimated based on the total annual precipitation as follows:

- For dry years with less than 190 mm total precipitation: runoff coefficient = 0.15.
- For average to wet years with more than 309 mm total precipitation: runoff coefficient = 0.30.
- Runoff coefficients for years with total precipitation between 190 mm and 309 mm: interpolated values between 0.15 and 0.30.

In 2016 (hydraulic year) the estimated total precipitation was 242 mm (Table 1), which corresponds to a modelled runoff coefficient of 0.22. The 2015 site-wide runoff coefficient, based on the 2015 water balance (measured flows, water inventory and total precipitation), is:

$$\textit{Annual Yield / Total Annual Precipitation} = \textit{Runoff Coefficient} \rightarrow \quad 60\ mm/243\ mm = \mathbf{0.25}$$

The calculated value for the annual site-wide runoff coefficient is slightly higher but close to the interpolated coefficient of 0.22. Overall, the agreement with site-wide runoff coefficients used for evaluating water management options is good and the model results can be expected to yield reliable estimates of the volume of water that must be managed on site on an annual basis.

Table 3: 2015 Water Inventory and Release to Minto Creek

Month/	MPTMF Volume Occupied (Water + Tailings)[A]	Change in MPTMF Water Inventory	Tailings Solids Deposition in MPTMF	A2PTMF Volume Occupied (Water + Tailings)[A]	Change in A2PTMF Water Inventory	Tailings Solids Deposition in A2PTMF	WSP Volume[A]	Change in WSP Water Inventory
Year	m^3	m^3/month	BCM/month	m^3	m^3/month	BCM/month	m^3	m^3/month
Jan-16	4,108,000	-81,000	0	1,024,000	83,000	45,000	82,000	-3,000
Feb-16	4,027,000	-175,000	0	1,152,000	147,000	38,000	79,000	-2,000
Mar-16	3,852,000	-211,000	0	1,337,000	160,000	47,000	77,000	10,000
Apr-16	3,641,000	90,000	0	1,544,000	42,000	44,000	87,000	18,000
May-16	3,731,000	-29,000	0	1,629,000	23,000	49,000	105,000	55,000
Jun-16	3,702,000	28,000	0	1,702,000	-5,000	41,000	160,000	18,000
Jul-16	3,730,000	-28,000	0	1,737,000	52,000	45,000	177,000	23,000
Aug-16	3,702,000	-44,000	0	1,835,000	54,000	47,000	200,000	7,000
Sep-16	3,658,000	90,000	0	1,936,000	13,000	44,000	207,000	-85,000
Oct-16	3,749,000	67,000	0	1,993,000	-4,000	45,000	122,000	-48,000
Nov-16	3,815,000	63,000	0	2,034,000	-18,000	42,000	74,000	-3,000
Dec-16	3,878,000	46,000	0	2,058,000	-18,000	37,000	70,000	-4,000
Jan-17	3,924,000			2,077,000			67,000	
SUM		**-183,000**	**-**		**529,000**	**524,000**		**-15,000**

Source: X:\01_SITES\Minto\1CM002.024_Water_Balance_Support\2016_Water_Balance_Update\2016 Water Balance Update REV00 KNK.xlsx

Notes:

A – on the first day of the month.

Table 4: Water Balance Summary of the Minto Mine Site, 2016 (Jan to Dec)

	Units	**Main Pit TMF**	**Area 2 Pit TMF**	**WSP**
Volume Change 2016 (water + tailings)	m^3	-183,000	1,053,000	-15,000
Tailings Deposited, total	BCM	-	524,000	-
Water Volume Change 2016	m^3	-183,000	529,000	-15,000
Estimated Groundwater Inflow	m^3	-	30,000	-
Total Water Inventory Increase in 2016	**m^3**	364,000		
Total Water Discharged to Minto Creek	m^3	254,000		
Total Site-Wide Yield in 2016	**m^3**	**617,000**		

Source: X:\01_SITES\Minto\1CM002.024_Water_Balance_Support\2016_Water_Balance_Update\2016 Water Balance Update REV00 KNK.xlsx

3 Water Quality Model Update

3.1 Solid Phase Geochemistry

The neutralization potential ratio (NPR) and copper content of waste rock and tailings were reviewed in order to identify any new trends in the solid phase geochemistry that may have developed since the last source term update. Significant changes in the solid phase geochemistry

would indicate a need for further analysis of the waste rock and tailings to generate new source terms that reflect the observed changes in the geochemistry.

The NPR and copper content of waste rock and tailings are shown in Figure 2 to Figure 5. No significant changes in geochemistry were observed in the properties of the materials produced in 2016 compared to similar materials produced in prior years. Therefore, no further evaluation of 2016 solid phase geochemistry was warranted.

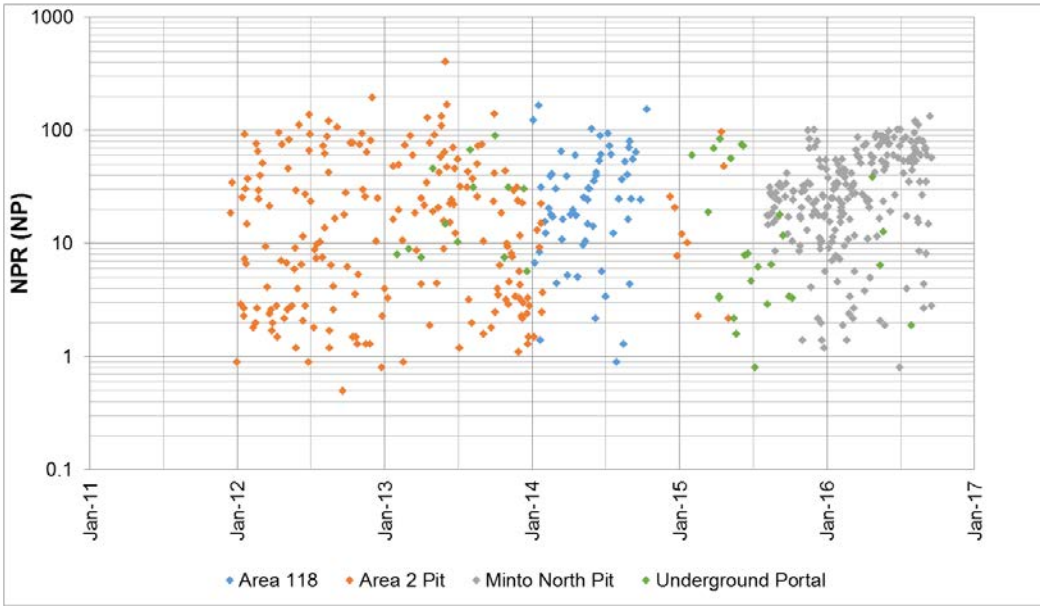


Figure 2: Waste Rock Neutralization Potential Ratio over Time

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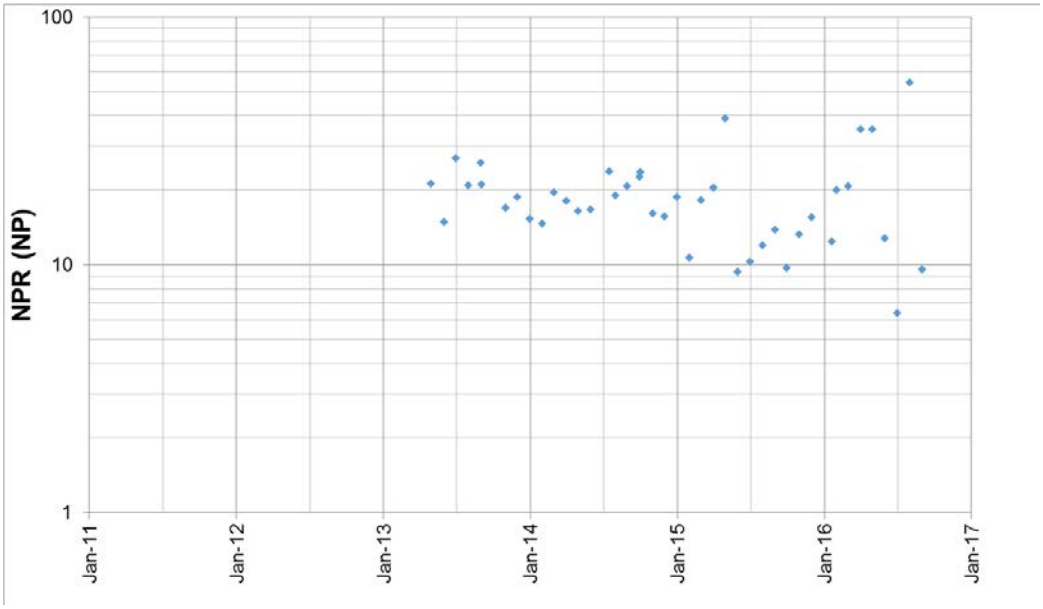


Figure 3: Tailings Neutralization Potential Ratio over Time

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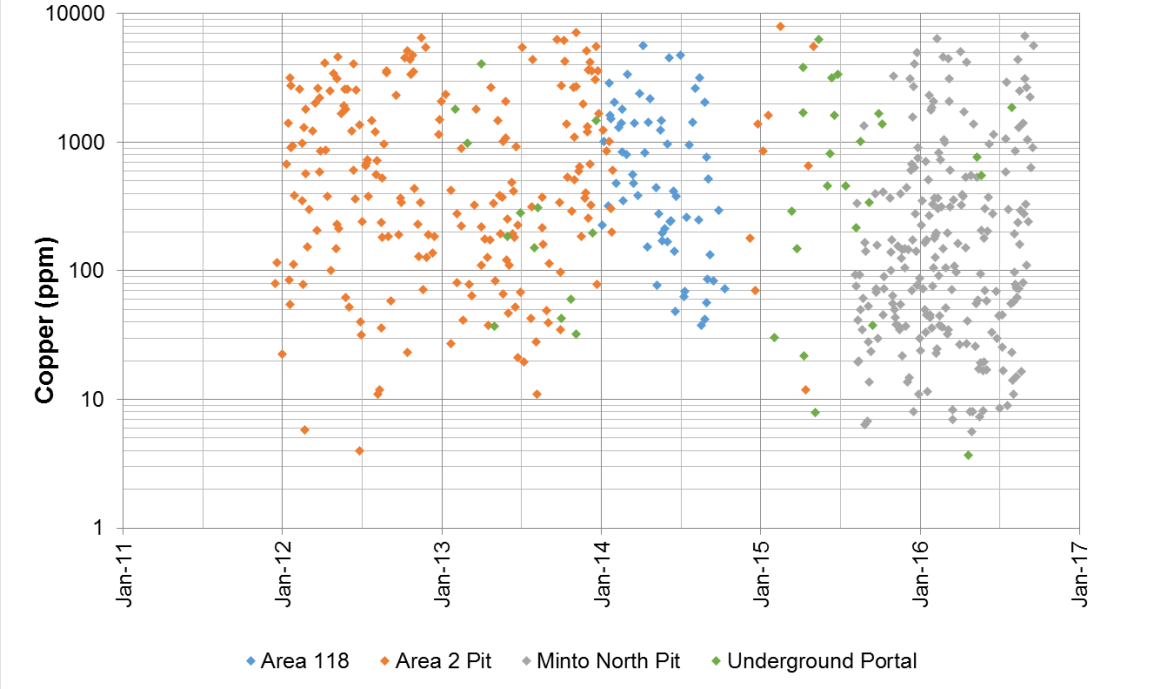


Figure 4: Waste Rock Copper Concentration over Time

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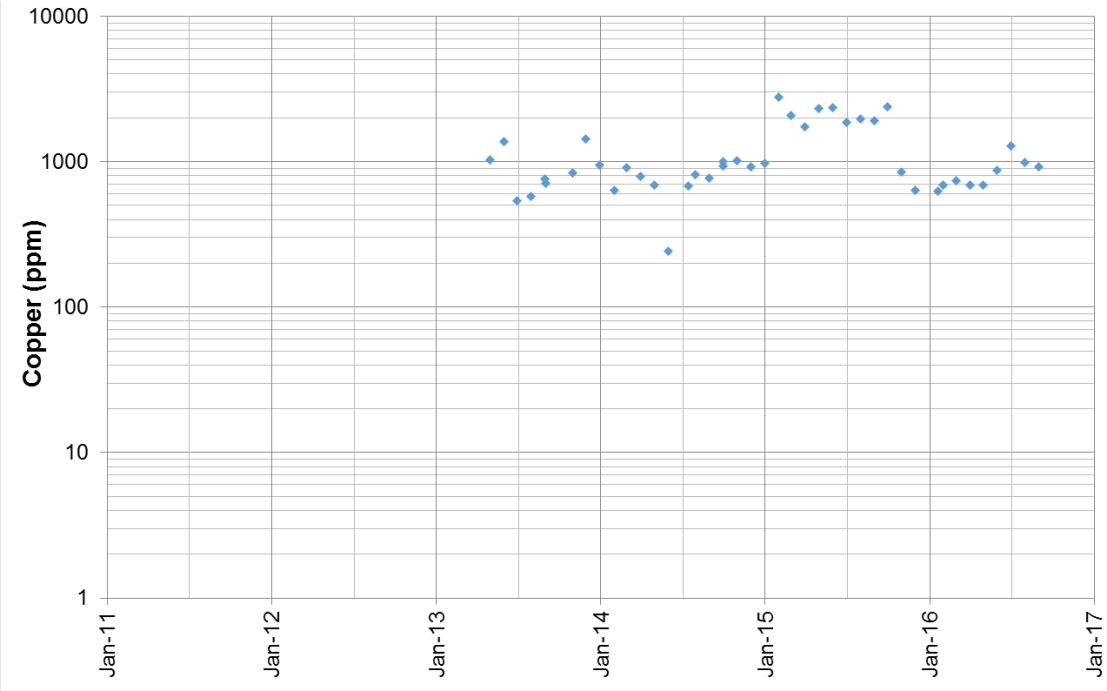


Figure 5: Tailings copper concentration over time

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3.2 Comparison of Measured Water Quality Data to Source Terms

3.2.1 Source Terms – Dry Stack Tailings Storage Facility

The Dry Stack Tailings Storage Facility (DSTSF) source terms used in the 2013 water and load balance model were developed based on the observed water chemistry at station W8. This station was chosen because it had the highest concentrations of copper, cadmium, selenium, and sulphate during the period of record available in 2013. The following points describe the source terms:

- Best Estimate source term: 50th percentile dissolved concentrations from W8 until 2013;
- Reasonable Worst Case source term: 95th percentile dissolved concentrations from W8 until 2013.

The Best Estimate and Reasonable Worst Case source terms were updated to include data to 2016 for comparison to the existing source terms. It was found that the updated source terms differed only marginally from the source terms used historically. Therefore, for consistency the 2013 DSTSF source terms will be used in the 2016 Updated Water Balance and Water Quality Model.

3.2.2 Source Terms – Waste Rock

The Main Waste Dump and the Southwest Waste Dump source terms used in the 2015 water and load balance model were defined based on the observed water chemistry at station W15, which collects runoff from the Southwest Waste Dump. Similar to the DSTSF source terms, the Best Estimate waste rock source term was the 50th percentile dissolved concentrations at W15 until 2015 and the Reasonable Worst Case waste rock source term was the 95th percentile dissolved concentrations from W15 until 2015.

The Best Estimate and Reasonable Worst Case source terms were updated to include data collected in 2016. Similar to the DSTSF source terms, the updated waste rock source terms differed only marginally from the source terms used historically. The 2015 waste rock source terms were therefore deemed acceptable and will be used in the 2016 Updated Water Balance and Water Quality Model for consistency.

3.2.3 Source Term – Tailings Slurry

A review of the mill influent water quality (reclaim water from the MPTMF, W12) and mill effluent water quality (Mill Thickener, W14) revealed that in 2015 and 2016 the processing of ore contributed greater loadings than in previous years. In light of this, the tailings slurry source term for the 2016 Updated Water Balance and Water Quality Model was updated as follows:

- Best Estimate source term: 50th percentile dissolved concentrations from W14 from 2015 and 2016 data only;
- Reasonable Worst Case source term: 95th percentile dissolved concentrations from W14 from 2015 and 2016 data only

However, the adjusted source terms for tailings slurry were not able to account for the greater-than-expected increase in parameter concentrations in the MPTMF. Similarly, adjustment of the waste rock source terms within limits of observed seepage concentrations were also not able to explain the concentration changes. One possible explanation for the additional observed load is that it is flushed from pit walls and waste rock as it is submerged by the rising water in the A2PTMF. This "first flush" effect is commonly observed and is caused by the buildup of weathering products as waste rock is exposed to the atmosphere for a period of time before becoming inundated. Once the A2PTMF has filled to capacity, this term is expected to become negligible. Ongoing monitoring and updates of the load balance will reveal whether flushing is indeed responsible for the additional loadings or if some other loading source not related to seepage or runoff characterized to date is the cause.

3.3 Water Quality Model Results

Table 6 and 7 show revised model outputs from the calibrated model of water quality in the Water Storage Pond (WSP) for 2016, 2018, and post-closure (best estimate and worst case) along with concentrations measured in 2016. Table 8 and 9 show revised model predictions of water quality in the MPTMF for 2016, 2018 and post-closure. Predictions for 2018 and post-closure were selected to provide representative short-term and long-term indications of water quality trends. Predictions are for average precipitation conditions. The Water Use Licence (QZ14-031) effluent limits are also listed in the tables. Model runs started on 1 January 2016 and ended on 1 January 2045, at which point concentrations in the model roughly have reached steady state.

The MPTMF was historically the primary water reservoir on site. In the model, the free water in the MPTMF and A2PTMF are more or less considered to belong to the same reservoir due to the high rate of flow between the two reservoirs. Reclaim water is drawn from the MPTMF and excess free water in the A2PTMF is pumped back to the MPTMF.

Therefore, a comparison of measured MPTMF water quality with concentrations predicted for pit water for the Phase V/VI environmental assessment provides a good measure of actual vs. expected geochemical performance of the site. Water collected in the WSP includes clean (non-contact) runoff and effluent from Minto's water treatment plant.

Median measured concentrations in the WSP in 2016 are comparable to the revised model predictions using best estimate source terms (Table 6). The favorable agreement indicates that the revised source terms are appropriate for describing the existing geochemical performance and the actual water management practices on site. Best estimate source terms are intended to provide an indication of the general trend in water quality parameter concentrations, but are not intended to capture maximum or outlier concentration values. Therefore, the median values of best estimate model predictions are compared to measured median values.

Revised model predictions using reasonable worst case source terms are generally higher than comparable median and maximum measured values for the WSP (Table 7).

Revised model predictions for water quality in the MPTMF (and by extension the A2PTMF) using the best estimate source terms are in good agreement with median measured concentrations in 2016 (Table 8) when the additional "first flush" load is included in the model. Water quality model predictions using reasonable worst case source terms are generally higher than measured

median and maximum concentrations, with the exception of dissolved copper. Median and maximum measured dissolved copper concentrations were both marginally higher than the concentrations predicted by the reasonable worst case source terms. The increase in dissolved copper concentration is not dramatic and may be caused by the flushing of rock that accompanied the inundation of the A2PTMF and related underground workings when deposition of tailings slurry was initiated. The source of the additional copper loadings (or increased copper solubility) will be evaluated based on water quality monitoring results in 2017.

Table 5: WSP Water Quality Model Predictions and Measured Concentrations in 2016, Best Estimate

		WUL Effluent Limits (QZ14-031)	WSP Measured Water Quality (Station W16)	Modelling Predictions of Quality in WSP (Station W16)		
Year			2016	2016	2018	Post-Closure
			Median	Median	Median	Median
Ammonia	mg/L	0.75	0.04	0.069	0.057	0.19
$N\text{-}NO_2$	mg/L	0.18	0.033	0.034	0.022	0.053
$N\text{-}NO_3$	mg/L	27.3	4.8	3.2	2	1.2
Ag-Dissolved	mg/L	0.0003	0.00001	0.000018	0.000013	0.000036
Al-Dissolved	mg/L	0.3	0.0066	0.034	0.032	0.14
As-Dissolved	mg/L	0.015	0.00034	0.00059	0.00048	0.0013
Cd-Dissolved	mg/L	0.0014[a]	0.0000075	0.000021	0.00002	0.000049
Cr-Dissolved	mg/L	0.003	0.0001	0.00093	0.00069	0.0015
Cu-Dissolved	mg/L	0.06/0.039[b]	0.018	0.012	0.0084	0.025
Fe-Dissolved	mg/L	3.3	0.027	0.37	0.31	0.49
Pb-Dissolved	mg/L	0.012	0.00005	0.00019	0.00015	0.00035
Mo-Dissolved	mg/L	0.219	0.0036	0.0021	0.0014	0.0076
Ni-Dissolved	mg/L	0.33	0.00066	0.0014	0.0012	0.002
Se-Dissolved	mg/L	0.006	0.0014	0.00084	0.00057	0.0025
Zn-Dissolved	mg/L	0.09	0.0013	0.005	0.0042	0.0081

Source: SRK, X:\01_SITES\Minto\1CM002.024_Water_Balance_Support\2015_Water_Balance_Update\All_Model_Results_for_WQ_Model_Comparison_for_2015_An_Report_SRJ_Rev00.xlsx

Notes:

Analytical data from Minto's water quality monitoring program.

a) at 50 mg/L hardness.

b) Cu effluent standard is 0.06 when [DOC] @ W2 > 10 mg/L and 0.039 when [DOC] @ W2 ≤ 10 mg/L.

Table 6: WSP Water Quality Model Predictions and Measured Concentrations in 2016, Reasonable Worst Case

Year		WUL Effluent Limits (QZ14-031)	WSP Measured Water Quality (Station W16)		Modelling Predictions of Quality in WSP (Station W16)					
			2016		2016		2018		Post-Closure	
			Median	Max	Median	Max	Median	Max	Median	Max
Ammonia	mg/L	0.75	0.04	0.31	0.16	0.23	0.11	0.21	0.31	0.41
N-NO_2	mg/L	0.18	0.033	0.094	0.17	0.26	0.11	0.24	0.1	0.13
N-NO_3	mg/L	27.3	4.8	5.7	20	30	12	29	3.4	4.5
Ag-Dissolved	mg/L	0.0003	0.00001	0.00002	0.000061	0.000089	0.00004	0.000085	0.00012	0.00019
Al-Dissolved	mg/L	0.3	0.0066	0.029	0.11	0.15	0.08	0.14	0.36	0.63
As-Dissolved	mg/L	0.015	0.00034	0.00088	0.00087	0.0011	0.00063	0.0011	0.0027	0.0047
Cd-Dissolved	mg/L	0.0014[a]	0.0000075	0.000017	0.000067	0.000095	0.000048	0.00009	0.00018	0.00029
Cr-Dissolved	mg/L	0.003	0.0001	0.001	0.0015	0.002	0.001	0.0019	0.0031	0.0049
Cu-Dissolved	mg/L	0.06/0.039[b]	0.018	0.026	0.029	0.043	0.019	0.041	0.062	0.089
Fe-Dissolved	mg/L	3.3	0.027	0.07	0.78	1.1	0.56	1	1.2	1.5
Pb-Dissolved	mg/L	0.012	0.00005	0.0002	0.00019	0.00025	0.00015	0.00024	0.001	0.0019
Mo-Dissolved	mg/L	0.219	0.0036	0.0051	0.0044	0.0063	0.0028	0.006	0.018	0.03
Ni-Dissolved	mg/L	0.33	0.00066	0.001	0.0024	0.0031	0.0017	0.003	0.0041	0.0058
Se-Dissolved	mg/L	0.006	0.0014	0.0016	0.004	0.006	0.0026	0.0057	0.0084	0.012
Zn-Dissolved	mg/L	0.09	0.0013	0.0063	0.0078	0.011	0.0059	0.01	0.016	0.025

Source: SRK, X:\01_SITES\Minto\1CM002.024_Water_Balance_Support\2015_Water_Balance_Update\All_Model_Results_for_WQ_Model_Comparison_for_2015_An_Report_SRJ_Rev00.xlsx

Notes:

Analytical data from Minto's water quality monitoring program.

a) at 50 mg/L hardness.

b) Cu effluent standard is 0.06 when [DOC] @ W2 > 10 mg/L and 0.039 when [DOC] @ W2 ≤ 10 mg/L.

Table 7: MPTMF Water Quality Model Predictions and Measured Concentrations in 2016, Best Estimate

		WUL Effluent Limits (QZ14-031)	MPTMF Measured Water Quality (Station W12)	Modelling Predictions of Quality in MPTMF (Station W12)		
Year			2016	2016	2018	Post-Closure
			Median	Median	Median	Median
Ammonia	mg/L	0.75	3.0	3.3	2.4	0.027
$N\text{-}NO_2$	mg/L	0.18	1.5	1.7	0.91	0.0057
$N\text{-}NO_3$	mg/L	27.3	26	22	18	0.14
Ag-Dissolved	mg/L	0.0003	0.00001	0.000037	0.000045	0.000045
Al-Dissolved	mg/L	0.3	0.011	0.13	0.18	0.15
As-Dissolved	mg/L	0.015	0.00045	0.0011	0.0015	0.0015
Cd-Dissolved	mg/L	0.0014[a]	0.000016	0.000038	0.000055	0.000048
Cr-Dissolved	mg/L	0.003	0.0001	0.0016	0.0018	0.002
Cu-Dissolved	mg/L	0.06/0.039[b]	0.025	0.017	0.029	0.031
Fe-Dissolved	mg/L	3.3	0.01	0.21	0.39	0.64
Pb-Dissolved	mg/L	0.012	0.00005	0.00035	0.00042	0.00043
Mo-Dissolved	mg/L	0.219	0.081	0.09	0.052	0.0061
Ni-Dissolved	mg/L	0.33	0.0017	0.0026	0.0025	0.0023
Se-Dissolved	mg/L	0.006	0.015	0.015	0.011	0.0023
Zn-Dissolved	mg/L	0.09	0.0037	0.008	0.0094	0.01

Source: SRK, X:\01_SITES\Minto\1CM002.024_Water_Balance_Support\2015_Water_Balance_Update\All_Model_Results_for_WQ_Model_Comparison_for_2015_An_Report_SRJ_Rev00.xlsx

Notes:

Analytical data from Minto's water quality monitoring program.

a) at 50 mg/L hardness.

b) Cu effluent standard is 0.06 when [DOC] @ W2 > 10 mg/L and 0.039 when [DOC] @ W2 ≤ 10 mg/L.

Table 8: MPTMF Water Quality Model Predictions and Measured Concentrations in 2016, Reasonable Worst Case

		WUL Effluent Limits (QZ14-031)	MPTMF Measured Water Quality (Station W12)		Modelling Predictions of Quality in MPTMF (Station W12)					
Year			2016		2016		2018		Post-Closure	
			Median	Max	Median	Max	Median	Max	Median	Max
Ammonia	mg/L	0.75	3	4.7	3.9	7.1	3.5	4.6	0.03	0.03
N-NO_2	mg/L	0.18	1.5	4.3	1.9	2.5	1.4	1.7	0.0099	0.011
N-NO_3	mg/L	27.3	26	29	34	45	45	49	0.65	0.78
Ag-Dissolved	mg/L	0.0003	0.00001	0.00002	0.00008	0.000096	0.00013	0.00014	0.00018	0.0002
Al-Dissolved	mg/L	0.3	0.011	0.018	0.27	0.34	0.42	0.44	0.43	0.46
As-Dissolved	mg/L	0.015	0.00045	0.0005	0.0021	0.0026	0.0031	0.0032	0.0031	0.0033
Cd-Dissolved	mg/L	0.0014[a]	0.000016	0.000034	0.00012	0.00014	0.0002	0.0002	0.00022	0.00024
Cr-Dissolved	mg/L	0.003	0.0001	0.001	0.0026	0.0031	0.0035	0.0038	0.0041	0.0045
Cu-Dissolved	mg/L	0.06/0.039[b]	0.025	0.053	0.039	0.05	0.073	0.078	0.082	0.091
Fe-Dissolved	mg/L	3.3	0.01	0.022	0.49	0.59	1	1.1	1.8	2
Pb-Dissolved	mg/L	0.012	0.00005	0.0002	0.00092	0.0011	0.0013	0.0013	0.001	0.0011
Mo-Dissolved	mg/L	0.219	0.081	0.087	0.098	0.13	0.071	0.087	0.018	0.019
Ni-Dissolved	mg/L	0.33	0.0017	0.0025	0.0038	0.0046	0.0049	0.0052	0.0055	0.006
Se-Dissolved	mg/L	0.006	0.015	0.017	0.019	0.028	0.019	0.022	0.011	0.013
Zn-Dissolved	mg/L	0.09	0.0037	0.0061	0.013	0.015	0.018	0.019	0.021	0.023

Source: SRK, X:\01_SITES\Minto\1CM002.024_Water_Balance_Support\2015_Water_Balance_Update\All_Model_Results_for_WQ_Model_Comparison_for_2015_An_Report_SRJ_Rev00.xlsx

Notes:

Analytical data from Minto's water quality monitoring program.

a) at 50 mg/L hardness.

b) Cu effluent standard is 0.06 when [DOC] @ W2 > 10 mg/L and 0.039 when [DOC] @ W2 ≤ 10 mg/L.

4 Closing

The summary of the 2016 Updated Water Balance and Water Quality Model for the Minto Mine was prepared in support of annual reporting. SRK would be pleased to address any questions or comments.

Appendix D: Pit Water Quality Analysis and Source Term Update – Minto Mine

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Memo

To:	Ryan Herbert, Minto Explorations Ltd.	**Client:**	Minto Explorations Inc.
From:	Kaitlyn Kooy	**Project No:**	1CM002.044
Cc:	Dylan MacGregor, Soren Jensen, SRK Consulting	**Date:**	February 7, 2018
Subject:	Pit Water Quality Analysis and Source Term Update- Minto Mine		

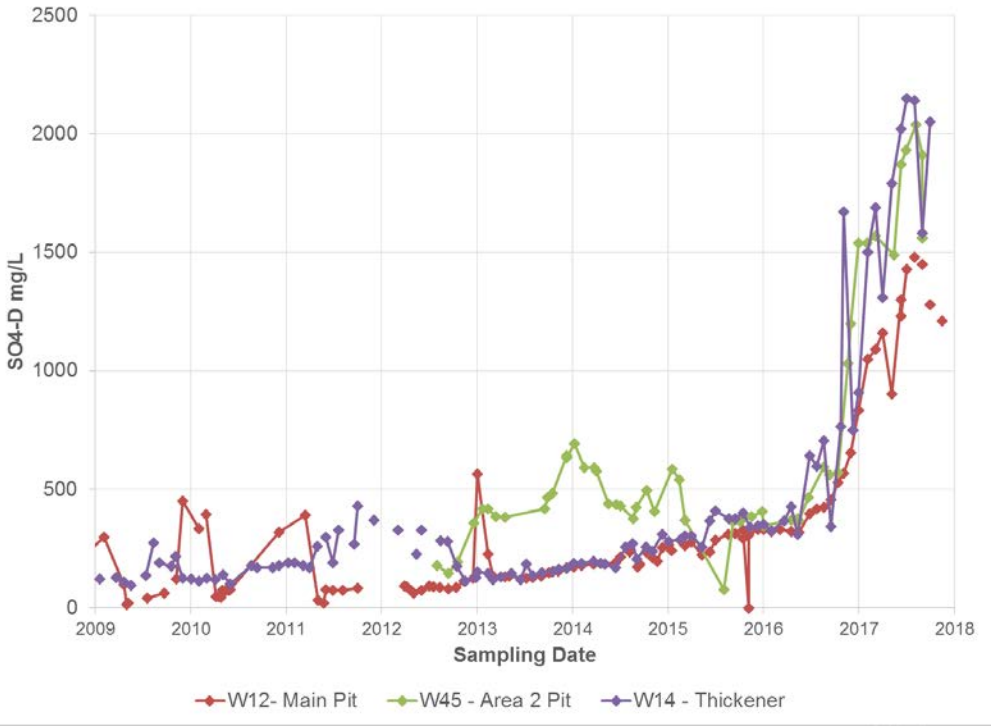
1 Introduction

Since the end of 2016, sulphate concentrations in the Main Pit and Area 2 Pit water have been increasing faster than observed previously. The resulting high sulphate concentrations have not been observed at any other surface source at the mine site and it has been unclear as to what has caused the observed sulphate concentrations. SRK was retained to review potential sources of sulphate at the Minto mine and to analyze the available data to determine a plausible explanation for the high concentrations of sulphate in the Main Pit and Area 2 Pit water. Results have contributed to the update of source terms used in the Water and Load Balance Model. This memo summarizes the methodology and results of that analysis and the source term update.

2 Background

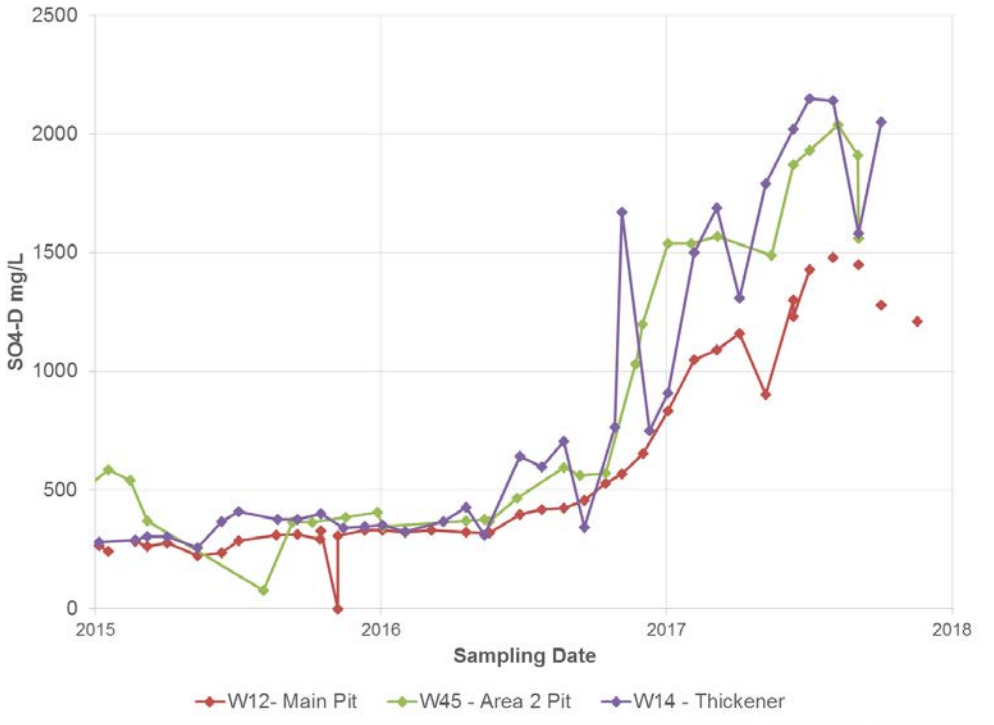
Steadily increasing sulphate concentrations have been observed in the Main Pit and Area 2 Pit water since the end of 2016. Both pits are Tailings Management Facilities (TMFs). Currently slurry tailings are pumped to the Area 2 Stage 2 Pit and reclaim water is drawn from the Main Pit free water. Water is actively pumped from Area 2 Pit to the Main Pit to maintain sufficient water levels in the Main Pit to meet the reclaim volume requirements. Since the Area 2 Pit became a TMF concentrations of most water quality constituents are higher in the Area 2 Pit than the Main Pit, as the Main Pit has a much larger catchment area and thus a much larger contribution of diluting waters. An aerial photograph taken in August of 2017 indicates that there is currently very little free water maintained in Area 2 as most of the free water is being pumped to the Main Pit.

Figure 1 shows the sulphate concentrations in both pits over time, and Figure 2 shows the sulphate concentrations in the pits over a shorter time scale (from 2015 to 2017). As shown in these figures, sulphate concentration in the pits has slowly increased from 2013 to mid-2016 before rising rapidly from the end of 2016 to the present day to a maximum of approximately 2000 mg/L in the Area 2 Pit and 1500 mg/L in the Main Pit. It appears as though sulphate levels in the Main Pit have decreased slightly in November and December of 2017 but it is unclear at this time if it will continue to trend downward.



Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\ML-ARD\Source_Terms_PhV_VI\Waste_Rock\SourceTerm_2017review\Pit Chemistry KNK WORKING\Raw Data Timeseries\MP_A2_Tails_UG_W15_TimeSeries_REV01_knk

Figure 1: Sulphate Concentrations in Main Pit, Area 2 Pit, and Tailings Thickener over time



Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\ML-ARD\Source_Terms_PhV_VI\Waste_Rock\SourceTerm_2017review\Pit Chemistry KNK WORKING\Raw Data Timeseries\MP_A2_Tails_UG_W15_TimeSeries_REV01_knk

Figure 2: Sulphate Concentrations in Main Pit, Area 2 Pit, and Tailings Thickener from 2015 to 2017

Since very little dilution occurs in the Area 2 Pit and the sulphate concentration is very similar to that of the tailings thickener, the Area 2 Pit sulphate loading was not directly analysed and is considered as a pathway only for sulphate loading in the Main Pit for the purposes of this analysis. The increase in sulphate load in the Main Pit is the primary focus of this analysis. Table 1 summarizes the annual sulphate loading in the Main Pit for the last three years.

Table 1: Sulphate Load in the Main Pit

Year	Sulphate Load in Main Pit at Year End (kg)[(1)]	Annual Load Increase (kg)
2015	437,324	135,694
2016	955,458	518,261
2017	1,601,671	639,825
	Total	**1,293,781**

Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\ML-ARD\Source_Terms_PhV_VI\Waste_Rock\SourceTerm_2017review\Pit Chemistry KNK WORKING\ Main_Pit_Sulphate_Review_Compliled_Calcs_Rev00_knk

Notes:

1. Includes sulphate load in free water only

3 Previous Source Terms

Past water quality predictions at the site have been based on source terms generated either by measured water quality at the site or laboratory humidity cell tests (e.g. SRK 2016). The source terms used in previous versions of the Water and Load Balance Model did not forecast the sulphate concentrations that are currently being observed in the open pits. In past modeling efforts in general, there were two groups of source term inputs that were modelled as reporting to the Main Pit and Area 2 Pit: contact water runoff and tailings slurry.

Table 2 below shows the modelled sulphate load reporting to the open pits annually from runoff based on the previous sulphate source terms (SRK 2016). The Best Estimate case is based on source terms generated from the 50th percentile of measured data while the Worst Case scenario is based on source terms generated from the 95th percentile of measured data.

Load from the tailings slurry is highly variable and is discussed further in Section 5.5.

Table 2: Typical Annual Main Pit Loading based on previous source terms

Catchment Type	Total Area Contributing Runoff to Open Pits (ha)	Annual Sulphate Runoff Load - Best Estimate (kg)	Annual Sulphate Runoff Load - Worst Case (kg)
Undisturbed	661	23,503	23,503(1)
Pit Walls	44	6,653	13,312
Dry Stack Tailings Storage Facility	19	2,285	3,808
Waste rock	196	43,039	157,401
Ore Stockpiles	13	1,735	1,788
Mill Area	16	870	1,740
Total	**950**	**78,085**	**201,552**

Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\ML-ARD\Source_Terms_PhV_VI\Waste_Rock\SourceTerm_2017review\Pit Chemistry KNK WORKING\ Main_Pit_Sulphate_Review_Compliled_Calcs_Rev00_knk

Notes:

1. The Best Estimate undisturbed source term is used for both the Best Estimate and Worst Case predictions

4 Sulphate Source Review: Pre-Screening

There are several possible sources of increased sulphate loading at the site. In order to determine the most likely source that is increasing sulphate loading in the system, all the potential sulphate sources were compiled and a screening process was carried out to eliminate sources that are unlikely to produce sulphate load of any appreciable magnitude. Table 3 summarizes all of the potential sources and whether or not they were carried forward for further evaluation. Figure 3 shows the simplified water routing assumptions used for the purposes of this analysis.

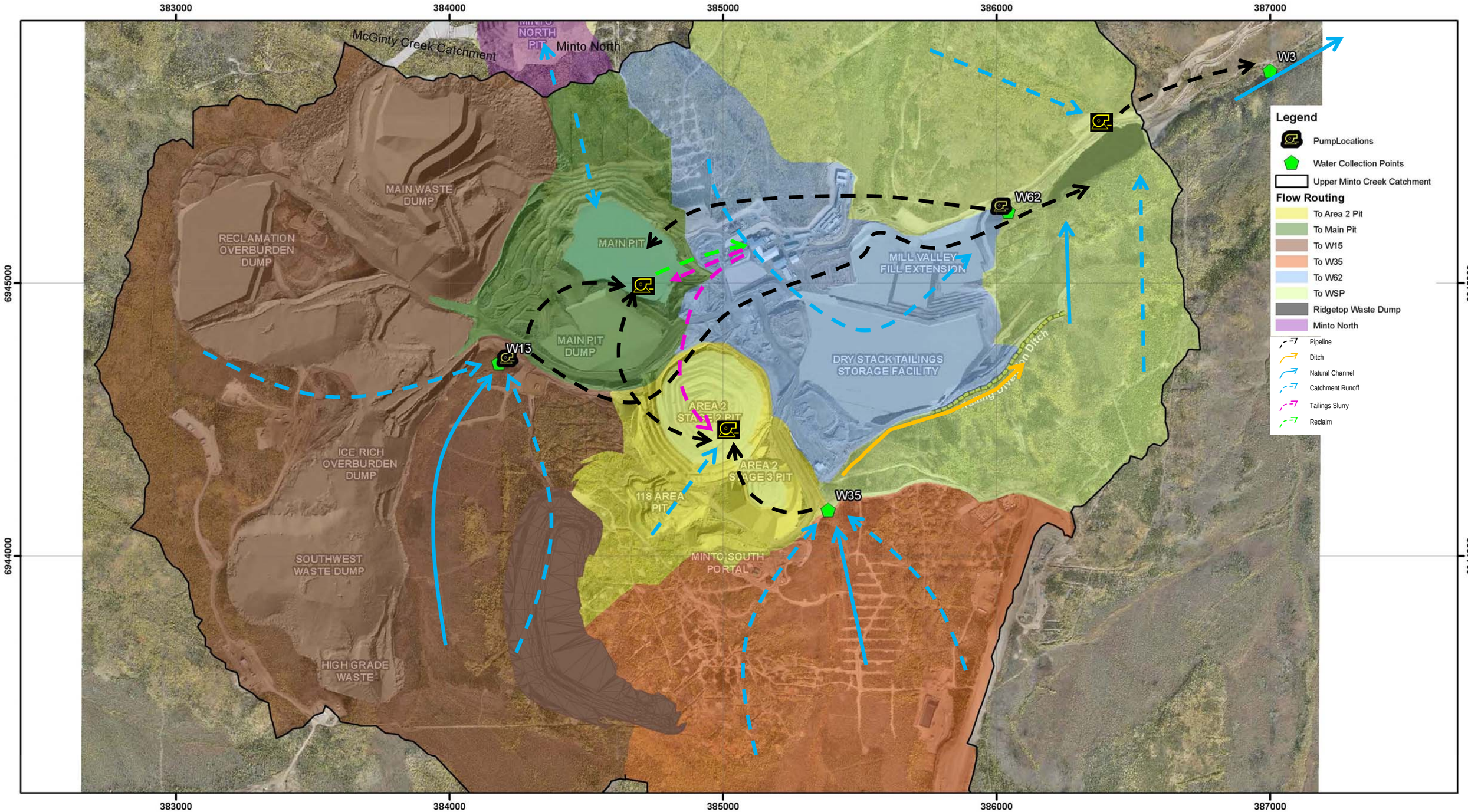
The sources carried forward from pre-screening are the Pit Walls, Main Pit SAT Dump, Underground Dewatering, Tailings Slurry, and Unsaturated Tailings.

Table 3: Potential Sulphate Source Pre-Screening Summary

Potential Sulphate Source	Water Quality Directly Monitored	Observed Increase in Sulphate Concentration	Carried Forward	Comment
Southwest Waste Dump	Yes	No	No	Measured water quality at station W15 and waste rock dump seepage monitoring points indicates no increase in sulphate levels
Main Pit SAT	No	N/A	**Yes**	Not currently monitored, cannot confirm sulphate loading from source has not increased
Dry Stack Tailings Storage Facility	Yes	No	No	Measured water quality at station W8A indicates no increase in sulphate levels
Unsaturated Tailings	No	N/A	**Yes**	Not currently monitored, cannot confirm sulphate loading from source has not increased
Pit Walls	No	N/A	**Yes**	Not currently monitored, cannot confirm sulphate loading from source has not increased
Tailings Slurry	Yes	Yes	**Yes**	Measured water quality in the tailings thickener (station W14) indicates an increase in sulphate concentration
Underground Mine	Yes	Yes	**Yes**	Measured water quality in the underground dewatering sump (station UG1) indicates an increase in sulphate concentration
Ore Stockpile	Yes	No	No	Measured water quality at ore stockpile seepage monitoring points indicates no increase in sulphate levels

Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\ML-ARD\Source_Terms_PhV_VI\Waste_Rock\SourceTerm_2017review\Pit Chemistry KNK WORKING\Main_Pit_Sulphate_Review_Compliled_Calcs_Rev00_knk

Figure 3: Water Routing Assumptions



Note: Adapted from Figure 4.2 in SRK (2018)

5 Potential Sulphate Source Evaluation

5.1 Main Pit SAT

The Main Pit SAT that is stored within the South Wall Buttress and overlying Main Pit Dump has been identified as a potential source because the dump contains waste rock with neutralization potential ration values less than three and there is currently no data available on the water quality of seepage from the dump.

The Main Pit SAT will be covered or submerged under water in the Main Pit at closure. There are currently no seepage monitoring locations (or opportunities) at the dump. There is no data available directly regarding the water quality of runoff from the dump and seepage from the dump may be a contributing factor to the higher than expected levels of sulphate in the open pits. The previous source terms regarding the water quality of the SAT seepage are based upon measured water quality at station W15.

Due to the lack of water quality data directly from the Main Pit SAT, the evaluation of the material as a potential sulphate source is conceptual in nature and has been undertaken in a conservative manner. In general, the evaluation methodology involved making estimates of maximum potential sulphate concentrations (based on industry experience at other mines) and then calculating the maximum possible sulphate load that may originate from the Main Pit SAT based on applying the mean annual precipitation (MAP) and assumed runoff coefficient to the assumed catchment area. A summary of the evaluation is shown in Table 4.

Table 4: Maximum Potential Load from Main Pit SAT Dump

Annual Runoff Volume (m^3)(1)	Maximum Potential Sulphate Concentration (mg/L)(2)	Maximum Annual Load (kg)
7,400	6,000	44,400

Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\ML-ARD\Source_Terms_PhV_VI\Waste_Rock\SourceTerm_2017review\Pit Chemistry KNK WORKING\Main_Pit_Sulphate_Review_Compliled_Calcs_Rev00_knk

Notes:

1. Based on a SAT catchment area of 7.45 ha, mean annual precipitation (MAP) of 329 mm and runoff coefficient of 0.3
2. Maximum potential sulphate concentration is not expected but is used to conservatively illustrate the scale of potential loading

In general it was concluded that the Main Pit SAT is unlikely to be the sole source of sulphate in the open pits since its potential annual maximum load is less than a tenth of the sulphate loading observed in the Main Pit over the first eight months of 2017. If a more in-depth refinement of this scoping evaluation is carried out in future, it is expected that the estimate of the maximum possible annual load would decrease. The volume of runoff that comes into contact with the dump is too small to carry the load required to reasonably explain the increase of sulphate in the Main Pit and Area 2 Pit.

5.2 Open Pit TMF Unsaturated Tailings

The Main Pit TMF and Area 2 TMF unsaturated tailings have been identified as a potential sulphate source because residual sulphide minerals contained in unsaturated tailings can weather over time and generate sulphate-bearing seepage, and there is currently no data available on the water quality of runoff from the unsaturated tailings beaches.

Based on the aerial photograph taken in August of 2017, the Main Pit currently has a relatively small tailings beach (approximately 11,000 m^2 as of August 2017) while the Area 2 Pit TMF has a relatively larger tailings beach (approximately 35,000 m^2 as of August 2017) since most of the free water in the Area 2 Pit is pumped to the Main Pit.

Similar to the Main Pit SAT evaluation, the unsaturated tailings evaluation remains conceptual and conservative due to the limited data available. The same methodology was used in this case by adopting estimates of maximum potential sulphate concentrations (based on industry experience at other mines) and calculating the maximum possible sulphate load that may originate from the unsaturated tailings in the Main Pit TMF and Area 2 Pit TMF. A summary of the calculations is shown in Table 5.

Table 5: Maximum Potential Plausible Load from Unsaturated Tailings in the Pit TMFs

Annual Main Pit and Area 2 Pit Tailings Beach Runoff Volume (m^3)(1)	Maximum Potential Sulphate Concentration (mg/L)(2)	Maximum Annual Load (kg)
4,600	6,000	27,600

Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\ML-ARD\Source_Terms_PhV_VI\Waste_Rock\SourceTerm_2017review\Pit Chemistry KNK WORKING\Main_Pit_Sulphate_Review_Compliled_CaIcs_Rev00_knk

Notes:

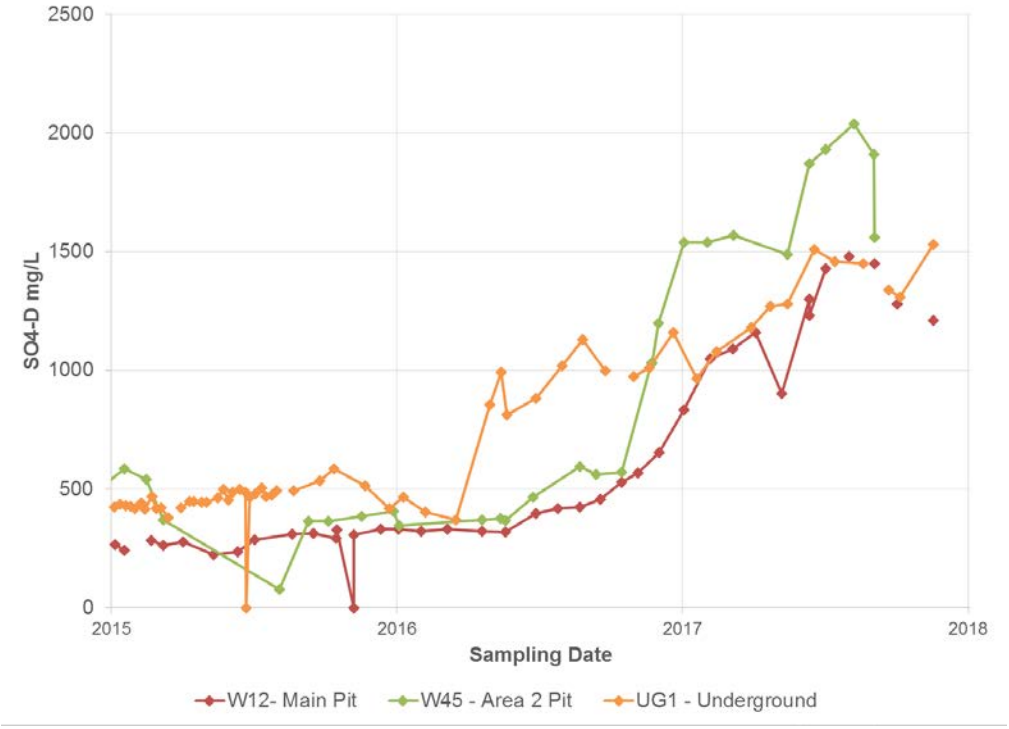
1. Based on the combined Main Pit and Area 2 Pit tailings beach catchment area of 4.6 ha approximated from the August 2017 aerial photo, mean annual precipitation (MAP) of 329 mm and runoff coefficient of 0.3
2. Maximum potential sulphate concentration is not expected but is used to illustrate the scale of potential loading

In general it was concluded that the unsaturated tailings is unlikely to be the sole source of sulphate in the open pits. The volume of runoff that comes into contact with the tailings beach is too small to carry the load required to reasonably explain the increase in sulphate concentrations in the Main Pit and Area 2 Pit. If a more in-depth refinement of this scoping evaluation is carried out in future, it is expected that the estimate of the maximum possible annual load would decrease.

5.3 Underground Mine

The underground mine dewatering was identified as a potential source because the observed water quality data had not previously been analysed in detail for this purpose, and sulphate concentrations in the underground mine discharge have increased markedly since mid-2016.

Water reporting to the underground workings is actively pumped or trucked to either the Main Pit or the Area 2 Pit. Water Quality is monitored monthly at the underground sump. Sulphate levels had remained relatively steady at approximately 500 mg/L prior to May 2016. After May 2016 sulphate concentrations in the underground mine have increased to a maximum of approximately 1500 mg/L. Figure 4 shows the sulphate concentrations in the underground mine over time as well as the sulphate concentrations in the Main Pit and Area 2 Pit for comparison.



Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\ML-ARD\Source_Terms_PhV_VI\Waste_Rock\SourceTerm_2017review\Pit Chemistry KNK WORKING\Raw Data Timeseries\MP_A2_Tails_UG_W15_TimeSeries_REV01_knk

Figure 4: Sulphate Concentrations in Main Pit, Area 2 Pit, and the Underground Mine from 2015 to 2017

The pumping rates from the underground mine sump is tracked, and that allowed the annual sulphate load conveyed from the underground mine to the pits to be calculated. The annual load from the underground is shown in Table 6.

Table 6: Total Water Volume and Sulphate Load Pumped from Underground from 2015 to 2017

Year	Annual Water Pumped (m^3)(1)	Annual Sulphate Load (kg)(2)
2015	39,492	19,035
2016	23,728	23,074
2017(3)	39,632	49,532
Total	**102,852**	**91,641**

Source: \\VAN-SVR0\Projects\01_SITES\Minto\!020_Site_Wide_Data\Water Management Data\Underground_Dewatering_Rev00_knk.xlsm

Notes:

1. Daily pumping rates derived from monthly averages and added together to calculate the annual total
2. Load calculated from interpolated sulphate concentration records
3. 2017 totals are up to and including data from August 8, 2017

The pumping rates from the underground to the pits are relatively low. A groundwater discharge to the pits from underground is unlikely as the underground workings are lower elevation than the pit bottoms, and groundwater modelling carried out on both the Main Pit and the Area 2 Pit indicates that there is a net seepage from the each pit to groundwater of tens of m^3/day (SRK 2018). It would be reasonable to conclude that there is no inflow of elevated sulphate groundwater seeping into the pits as the net groundwater flux is outward seepage from the open pits.

Is it important to note that the increase in sulphate in the underground mine occurred prior to the increase in sulphate in either of the open pits. This timing relationship makes it unlikely that the elevated levels of sulphate in the pits could have caused the elevated sulphate levels in the underground mine.

In general is was concluded that the underground mine discharge is a relatively small source of sulphate in the system, and cannot account for all of the sulphate inventory increase in the open pits.

5.4 Open Pit Walls

The open pit walls have been identified as a potential source of sulphate loading because there is currently no data available on the water quality of runoff from the pit walls. The current assumptions regarding the water quality of the pit walls runoff are based upon laboratory humidity cell tests that had been carried out in 2013.

Due to the lack of water quality data from the pit walls runoff, the evaluation of the pit walls as a potential sulphate source is conceptual and conservative in nature. In general, the evaluation methodology involved making estimates of maximum potential sulphate concentrations (based on the 95th percentile values observed in Minto waste rock seepage) and then calculating the maximum possible sulphate load that may originate from the pit walls based on the mean annual precipitation (MAP) and assumed runoff coefficient. A summary of the evaluation is shown in Table 7.

Table 7: Maximum Potential Load from Open Pit Walls

Annual Main Pit and Area 2 Pit Runoff Volume (m^3)(1)	Maximum Potential Sulphate Concentration (mg/L)(2)	Maximum Annual Load (kg)
32,400	600	19,440

Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\ML-ARD\Source_Terms_PhV_VI\Waste_Rock\SourceTerm_2017review\Pit Chemistry KNK WORKING\ Main_Pit_Sulphate_Review_Compliled_Calcs_Rev00_knk

Notes:

1. Based on a mean annual precipitation (MAP) of 329 mm and runoff coefficient of 0.3
2. Maximum potential sulphate concentration is not expected but is used to illustrate the scale of potential loading

In general it was concluded that the open pit walls are unlikely to be the sole source of sulphate in the open pits since the maximum possible annual sulphate load is less than a twentieth of the sulphate loading observed in the Main Pit in 2017. The volume of runoff that comes into contact

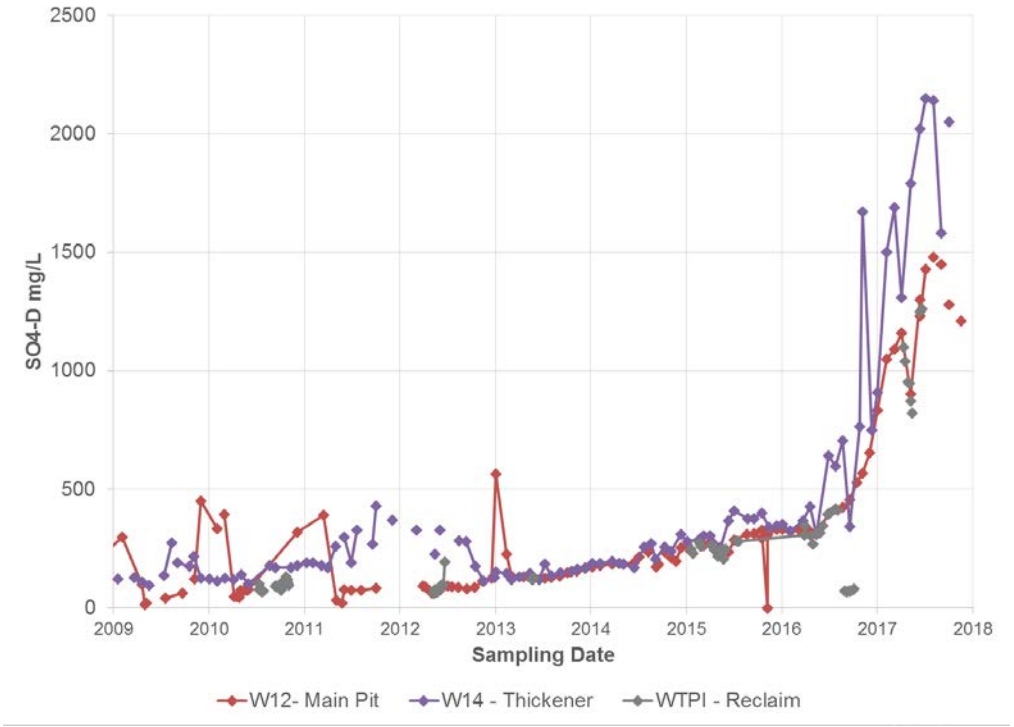
with the pit walls is too small to carry the load required to reasonably explain the increase of sulphate in the Main Pit and Area 2 Pit.

5.5 Tailings Slurry

The tailings slurry has been identified as a potential source of sulphate loading because the concentration of sulphate in the tailings thickener has increased significantly since mid-2016 and the tailings slurry water quality significantly influences the water quality in the open pits due to the relatively high volumes of tailings slurry pumped out of the mill.

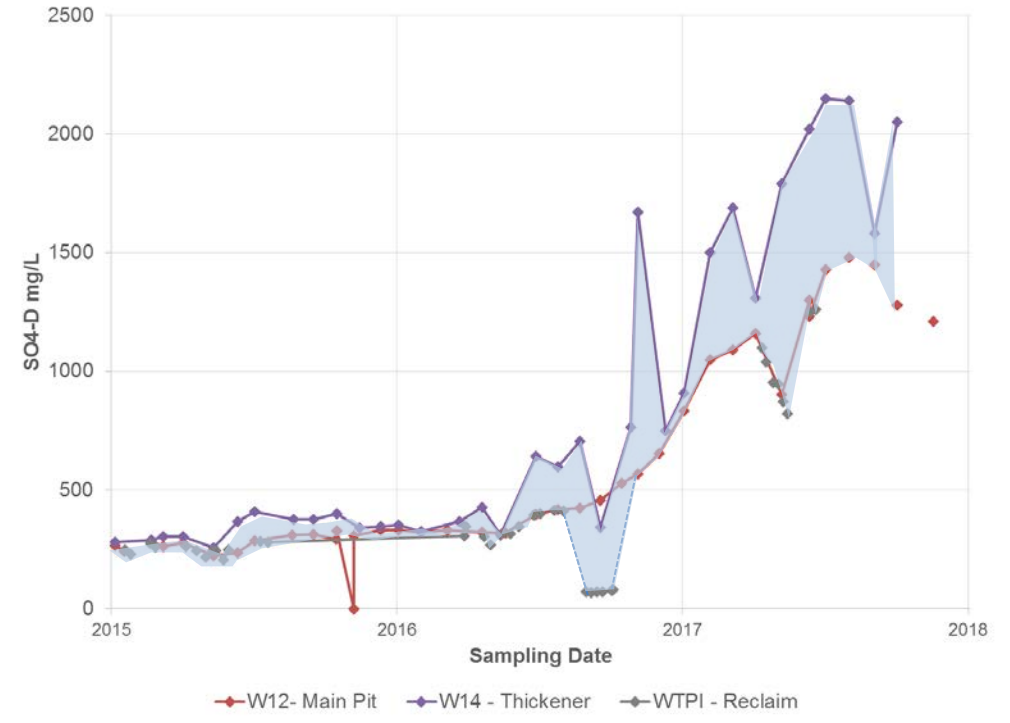
5.5.1 Mill Process Sulphate Loading

Figure 5 shows the sulphate concentration in the Main Pit, Tailings Thickener, and the process water feed tank over time. Mill process water is typically pulled from the Main Pit but there was a brief period of time in the fall of 2016 where treated water from the water treatment plant effluent was used as reclaim water. Figure 6 shows the tailings thickener and reclaim sulphate concentrations from 2015 to 2017 and demonstrates that the difference between the two has increased significantly in 2016 and 2017. The blue shading in Figure 6 highlights this difference and represents the sulphate that is added to the system from the mill.



Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\ML-ARD\Source_Terms_PhV_VI\Waste_Rock\SourceTerm_2017review\Pit Chemistry KNK WORKING\Raw Data Timeseries\MP_A2_Tails_UG_W15_TimeSeries_REV01_knk

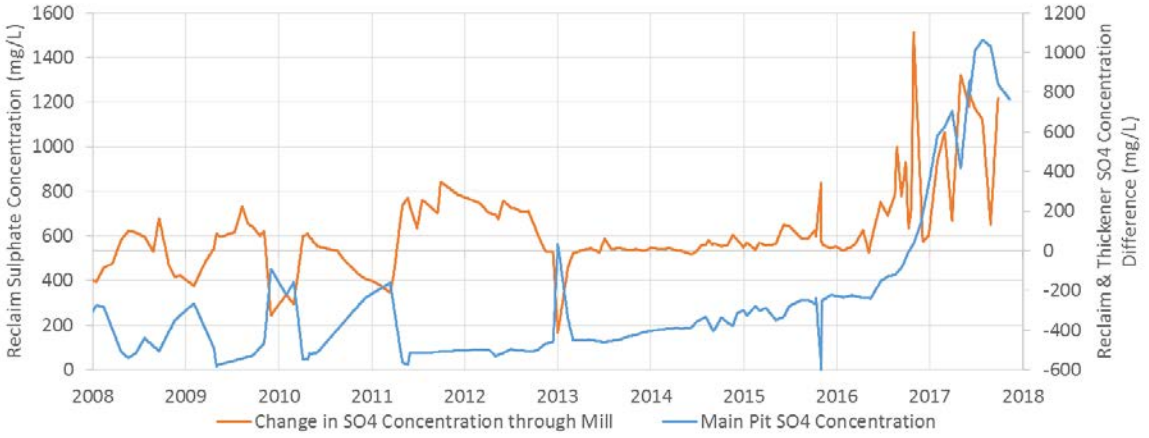
Figure 5: Sulphate Concentrations in Main Pit, Tailings Thickener, and Reclaim Water over time



Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\ML-ARD\Source_Terms_PhV_VI\Waste_Rock\SourceTerm_2017review\Pit Chemistry KNK WORKING\Raw Data Timeseries\MP_A2_Tails_UG_W15_TimeSeries_REV01_knk

Figure 6: Sulphate Concentrations in Main Pit, Tailings Thickener, and Reclaim Water from 2015 to 2017

Figure 7 demonstrates that the higher loading rates in the mill (orange line) coincide with the increase in sulphate concentration in the Main Pit (blue line). The sulphate loading rate in the mill was never previously as high as it was in late 2016 and 2017. Table 8 shows the total additional sulphate load to free water from the tailings slurry from 2015 to 2017.



Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\ML-ARD\Source_Terms_PhV_VI\Waste_Rock\SourceTerm_2017review\Pit Chemistry KNK WORKING\Main_Pit_Mass_Balance_Rev01_knk

Figure 7: Sulphate Concentrations in Reclaim Water and Concentration Change in Mill Tailings Process Water over Time

Table 8: Annual Sulphate Loading to Pits from the Mill since 2015

Year	Annual Sulphate Load Increase in Mill Process (kg)[1]	Assumed load Increase to Tailings Pores (kg)[2]	Annual Sulphate Load Increase to Free Water (kg)[3]
2015	135,457	40,490	94,966
2016	398,068	156,335	241,732
2017[4]	588,413	219,654	368,759

Notes:

1. Difference between sulphate load in Main Pit reclaim water and thickener discharge sulphate load
2. Based on the proportion of water that is lost to tailings pores
3. Based on remaining water volume that reports to open water pond (i.e. is not lost to pores)
4. Up to and including August 2017

5.5.2 Sulphide Ore & POX Blend

The type and grade of ore processed in the mill is variable over time. The ore is classified as either sulphide ore or POX (partially oxidized) ore. The degree of oxidation of POX ore is variable and has a variable effect on the water quality of the mill effluent. Figure 8 shows the type of ore milled from 2015 to 2017. Sulphide ore was milled from January to October of 2016 and a blend of sulphide ore and POX was milled in 2015 and from November 2016 to present.

Typically, the milling of purely sulphide ore does not result in high sulphate loading which is evidenced by the record of low sulphate concentrations in the mill effluent. The milling of POX, depending on the grade, can periodically have a significant influence on the water quality of the mill effluent as shown by the 2017 sulphate loading rates.

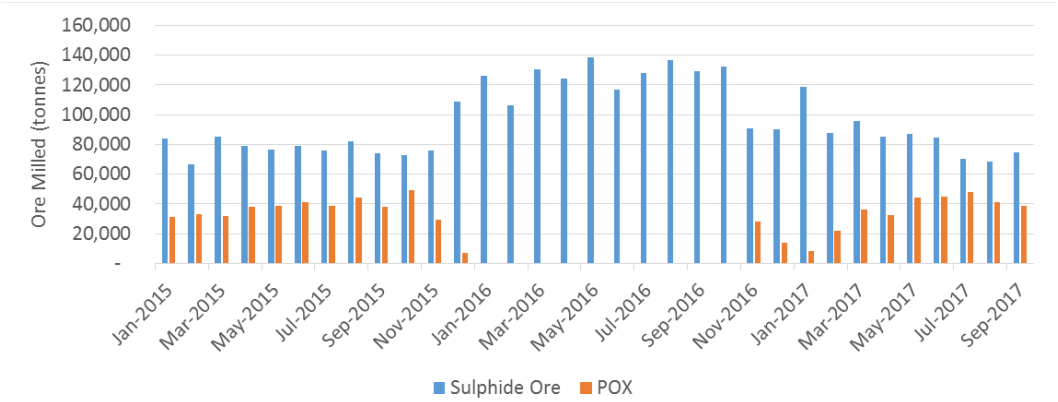


Figure 8: Ore Type Milled from 2015 to 2017

When POX ore is being milled, sodium sulphide is used in the mill process. Figure 9 shows the sodium sulphide use in the mill from 2015 to 2017. Sodium sulphide introduces sulphide ions into the system which eventually oxidize to sulphate in surface water, and therefore represents an imported source of sulphur to the system. However, until that oxidation occurs, the sulphur remains in sulphide form- that sulphur would be reported in a total sulphur analysis, but not in a routine sulphate analysis.

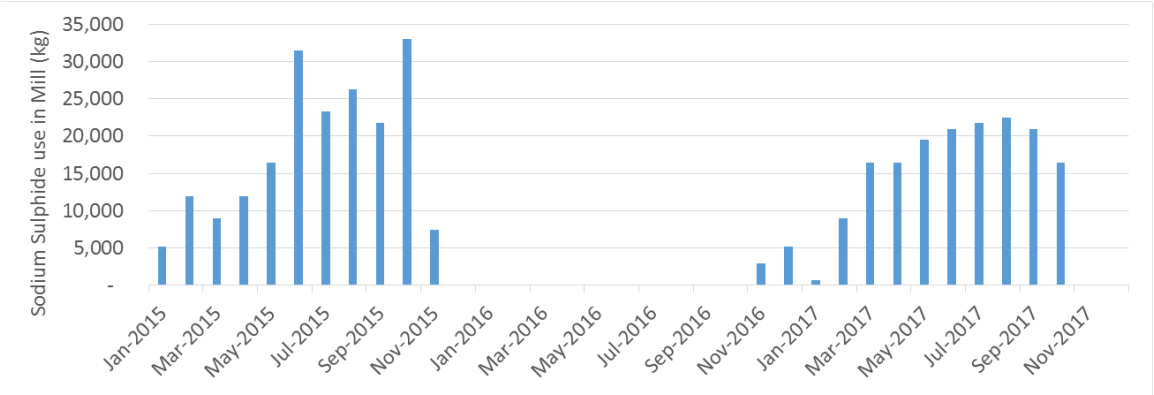


Figure 9: Sodium Sulphide use in the Mill from 2015 to 2017

The introduction of sodium sulphide into the process system in 2015 did not result in any observed increase in sulphate loading in the mill while sodium sulphide use in 2017 coincides with a significant increase in observed sulphate loading. This suggests that the POX being milled in 2015 and 2017 had very different soluble sulphate contents and have responded quite differently to the mill process and addition of sodium sulphide. This is further evidenced by the

higher levels of unoxidized total sulphur observed in the tailings thickener, Area 2 Pit, and Main Pit in late 2016 and 2017. Typically the total sulphur observed in the open pits and tailings thickener is nearly entirely in the form of sulphate. However, in late 2016 and 2017 levels of unoxidized sulphur are periodically elevated higher than have been observed previously. Figure 10 shows the concentration of unoxidized sulphur in the tailings thickener, Area 2 Pit, and Main Pit over time. Equation 1 shows the method for calculating unoxidized sulphur.

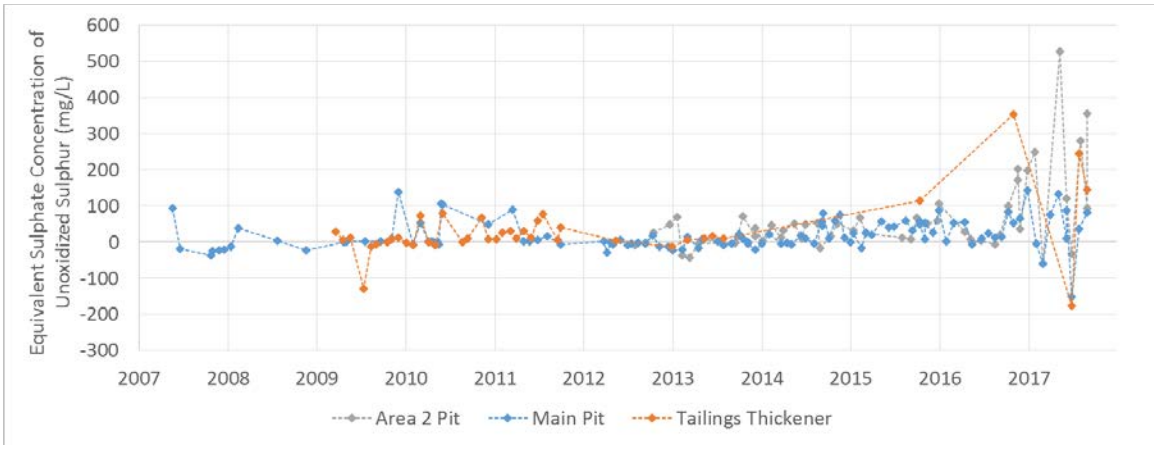


Figure 10: Unoxidized Sulphur Converted to Sulphate Equivalent

Equation 1: Unoxidized sulphur calculation.

$$Sulphur\ Concentration\ x\ \frac{Sulphate\ molar\ mass}{Sulphur\ molar\ mass} - Sulphate\ Concentration = Unoxidized\ Sulphur$$

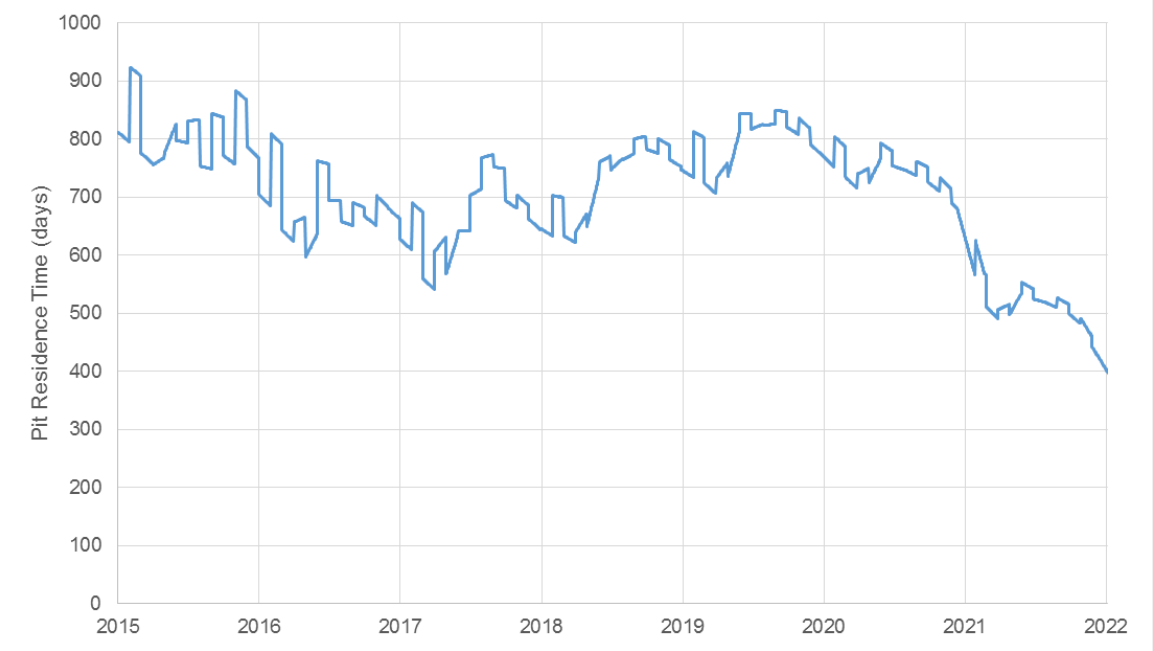
Unoxidized sulphur appears to decline as the water is pumped from the mill to the Area 2 Pit to the Main Pit. This indicates that there is further additional sulphate generation occurring in the system as sulphur oxidizes. The total sulphur concentration data at the tailings thickener is somewhat limited so further review of the data may provide additional insight as more data becomes available.

The contrast in unoxidized sulphur and sulphate concentration observed in the tailings thickener and open pits in 2015 and 2017, both years where POX was milled, speaks to the variability of the mill ore feed and inherent difficulty in predicting the mill effluent water quality.

5.5.3 Mill Reclaim Residence Time

Currently mill reclaim water is sourced from the Main Pit, slurry tailings are deposited in the Area 2 Pit, and the excess free water in the Area 2 Pit is pumped back to the Main Pit. This circular system has a residence time that is a function of the available free water in the two open pits. Based on the water level record in the open pits, the tailings production schedule, and modelled runoff water volumes, the mill reclaim system residence time has been steadily decreasing since 2015 as water is lost to tailings pores and subaqueous waste rock pores. Figure 11 shows the approximate predicted residence time in the open pit ponds during operations. The available free water and subsequently the residence time in the pits appears to reach a local minimum in early

2017 and then begins to climb. The predicted increase in residence time immediately following the minimum is due to the overall volume in the open pits increasing due to accumulation of runoff and minimal discharge of any water from the system.



Source: Z:\01_SITES\Minto\!020_Site_Wide_Data\ML-ARD\Source_Terms_PhV_VI\Waste_Rock\SourceTerm_2017review\Pit Chemistry KNK WORKING\Pit_Source_Proportions_Initial_MP_knk

Figure 11: Predicted Mill Reclaim/TMF System Residence Time

The predicted minimum residence time approximately coincides with the maximum observed sulphate concentration in the Main Pit and Area 2 Pit. The observed sulphate concentration in the open pits is beginning to trend downward as the predicted residence time in the open pits begins to increase. The effects of the higher mill sulphate loading rates observed recently have likely been pronounced by the relatively low pit residence time.

In general it was concluded that the tailings slurry is a significant source of sulphate at the Minto Mine and is highly sensitive to the ore feed and reagents used in the mill process. Water management and residence time in the open pits also has an effect on the water quality

6 Source Term Update

The underground and tailings slurry source terms have been updated to reflect the findings of the analysis and observed current conditions. The other potential sulphate sources reviewed as part of this analysis will not require a source term update as there is no measured data to substantiate any change from previous assumptions.

6.1 Underground Source Term

The underground source terms were generated as follows:

- **South Portal Best Estimate** – 50th Percentile of available water quality data from the underground sump (stations UG1 and W44); applies to Area 118 Underground, Area 2 Underground, and Copper Keel Underground
- **South Portal Worst Case** – 95th Percentile of available water quality data from the underground sump (stations UG1 and W44); applies to Area 118 Underground, Area 2 Underground, and Copper Keel Underground
- **Minto East Best Estimate** – 50th Percentile of available water quality data groundwater monitoring wells with observed elevated sulphate concentrations (stations MW12-05-01, MW12-05-02, MW12-05-03, MW17-12-01, and MW17-12-02); applies to Minto East Underground, Minto East 2 Underground, and Minto North Underground
- **Minto East Worst Case** – 95th Percentile of available water quality data groundwater monitoring wells with observed elevated sulphate concentrations (stations MW12-05-01, MW12-05-02, MW12-05-03, MW17-12-01, and MW17-12-02); applies to Minto East Underground, Minto East 2 Underground, and Minto North Underground

Since the water quality in the groundwater monitoring wells used to generate the Minto East underground source terms are not affected by active underground mining (i.e. do not have nitrogen species concentrations originating from explosive residues) the ammonia, nitrate, and nitrite South Portal source term values were also used for the Minto East underground source term. The underground source terms are provided in Attachment 1.

6.2 Tailings Slurry Source Term

The tailings slurry source terms were generated as follows:

- **Sulphide Ore Source Term Best Estimate** – 50th percentile of monthly average load generated per kg of sulphide ore milled from January to October 2016 (no POX milled during this period)
- **Sulphide Ore Source Term Worst Case** – 95th percentile of monthly average load generated per kg of sulphide ore milled from January to October 2016 (no POX milled during this period)
- **POX Low Sulphate Source Term Best Estimate** – 50th percentile of monthly average load generated per kg of POX ore milled in 2015
- **POX Low Sulphate Source Term Worst Case** – 95th percentile of monthly average load generated per kg of POX ore milled in 2015
- **POX High Sulphate Source Term Best Estimate** – 50th percentile of monthly average load generated per kg of POX ore milled from November 2016 to September 2017
- **POX High Sulphate Source Term Worst Case** – 95th percentile of monthly average load generated per kg of POX ore milled from November 2016 to September 2017

The tailings slurry source terms are provided in Attachment 2.

7 Discussion

The review has identified that the source of the increased sulphate concentration in the Main Pit and Area 2 Pit water is primarily soluble sulphate loadings from POX ore being processed in the mill, and the use of sodium sulphide to process the POX ore. Underground dewatering is also a source of sulphate loading but to a comparatively lesser extent presently; loadings from underground dewatering will increase over the period of mining in Minto East and Minto East 2. The revised sulphide ore, POX ore and underground discharge source terms presented herein should be incorporated into the next update to the Water and Load Balance.

SRK Consulting (Canada) Inc.



Kaitlyn Kooy, EIT
Staff Consultant

reviewed by



Dylan MacGregor, PGeo
Principal Consultant

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The opinions expressed in this report have been based on the information available to SRK at the time of preparation. SRK has exercised all due care in reviewing information supplied by others for use on this project. Whilst SRK has compared key supplied data with expected values, the accuracy of the results and conclusions from the review are entirely reliant on the accuracy and completeness of the supplied data. SRK does not accept responsibility for any errors or omissions in the supplied information, except to the extent that SRK was hired to verify the data.

8 References

SRK Consulting 2016. Water and Load Balance Model Report 2016. Prepared for Minto Explorations Ltd. August 2016.

SRK Consulting 2018. Minto Groundwater Characterization and Hydrogeologic Conceptual Model Update Report. Prepared for Minto Explorations Ltd. January 2018.

Attachment 1: Underground Source Terms

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Memo

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Cc:	Dylan MacGregor, Soren Jensen, SRK Consulting	**Date:**	February 5, 2018
Subject:	Underground discharge source terms: 2018 update		

1 Cover

Updated source terms for the underground discharge were developed in 2018, as described in SRK (2018).

> SRK Consulting 2018. Pit Water Quality Analysis and Source Term Update. Prepared for Minto Explorations Ltd. February 2018

South Portal Underground Discharge Source Term

	Expected Case	Worst Case
Parameter	**mg/L**	**mg/L**
Ag-D	0.00002	0.0000306
Al-D	0.0106	0.0242
Ammonia	18	91.4
As-D	0.00156	0.00426
Cd-D	0.0000165	0.000275
Cr-D	0.001	0.001
Cu-D	0.00742	0.0673
Fe-D	0.012	0.0758
Mn-D	0.132	0.205
Mo-D	0.0176	0.0419
Ni-D	0.0023	0.00817
Nitrate	35.1	181
Nitrite	1.44	7.65
Se-D	0.000475	0.00441
Sulphate	366	1170
Zn-D	0.00605	0.0583

Minto East Underground Discharge Source Term

	Site Monitoring Data	
	Expected Case	**Worst Case**
Parameter	**mg/L**	**mg/L**
Ag-D	0.00002	0.00002
Al-D	0.0045	0.016
Ammonia*	18	91.4
As-D	0.0005	0.00161
Cd-D	0.00001	0.0000235
Cr-D	0.000965	0.001
Cu-D	0.0002	0.00313
Fe-D	0.0395	3.04
Mn-D	0.713	2.6
Mo-D	0.001	0.00695
Ni-D	0.001	0.00112
Nitrate*	35.1	181
Nitrite*	1.44	7.65
Se-D	0.000101	0.000981
Sulphate	795	1160
Zn-D	0.005	0.0269

*Nitrogen species are based on South Portal water quality

Attachment 2: Tailings Slurry Source Terms

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Memo

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Cc:	Dylan MacGregor, Soren Jensen, SRK Consulting	**Date:**	February 7, 2018
Subject:	Tailings slurry source terms: 2018 update		

1 Cover

Updated source terms for the tailings slurry were developed in 2018, as described in SRK (2018).

> SRK Consulting 2018. Pit Water Quality Analysis and Source Term Update. Prepared for Minto Explorations Ltd. February 2018

Tailings Slurry Sulphide Ore Source Terms

	Expected Case	Worst Case
Parameter	**mg/kg**	**mg/kg**
Ag-D	0.0000041	0.00000819
Al-D	0.0159	0.0663
Ammonia	1.53	2.77
As-D	0.000172	0.000188
Cd-D	0.00000338	0.00000522
Cr-D	0.000041	0.00041
Cu-D	0.000463	0.00087
Fe-D	0.0041	0.0041
Mn-D	0.0382	0.0649
Mo-D	0.0399	0.0472
Ni-D	0.000606	0.00112
Nitrate	16.2	22.
Nitrite	0.592	1.45
Se-D	0.0074	0.00933
Sulphate	163.	302.
Zn-D	0.00041	0.00205

Tailings Slurry Low Sulphate POX Ore Source Terms

	Expected Case	Worst Case
Parameter	**mg/kg**	**mg/kg**
Ag-D	0.0000163	0.0000413
Al-D	0.101	0.194
Ammonia	6.82	15.7
As-D	0.000565	0.00252
Cd-D	0.0000151	0.0000563
Cr-D	0.00122	0.00372
Cu-D	0.00477	0.055
Fe-D	0.0149	0.0659
Mn-D	0.144	0.771
Mo-D	0.166	0.425
Ni-D	0.00159	0.0053
Nitrate	39.3	129.
Nitrite	2.02	3.7
Se-D	0.0356	0.079
Sulphate	438.	1280.
Zn-D	0.0062	0.0186

Tailings Slurry High Sulphate POX Ore Source Terms

	Expected Case	Worst Case
Parameter	**mg/kg**	**mg/kg**
Ag-D	0.0000159	0.0000202
Al-D	0.0359	0.0967
Ammonia	6.3	20.
As-D	0.000489	0.0018
Cd-D	0.0000741	0.000157
Cr-D	0.000243	0.000479
Cu-D	0.00523	0.0324
Fe-D	0.0243	0.0492
Mn-D	1.41	2.29
Mo-D	0.0974	0.0974
Ni-D	0.00325	0.00325
Nitrate	52.	149.
Nitrite	3.83	3.83
Se-D	0.0325	0.0325
Sulphate	2340.	4320.
Zn-D	0.00243	0.00661



Minto Mine

Operations Adaptive Management Plan

2018-01

Prepared by:
Minto Explorations Ltd.
Minto Mine
January 2018

Minto Mine Operations Adaptive Management Plan

First Issue: June 2014
Version History

Version Number	Issue Date	Description and Revisions Made
2014-01	June 2014	First Issue
2015-01	December 2015	Revisions made as per requirements of WUL QZ14-031 and QML-0001
2016-01	May 2016	Revisions made as per AMP comments from the Yukon Government, Selkirk First Nation and QML-0001 conditions
2017-01	January 2017	Revisions made as per Yukon Water Board and Yukon Government review comments.
2017-02	April 2017	Revisions made as per Yukon Water Board Reasons for Decisions (April 3, 2017).
2018-01	January 2018	Revised thresholds in Minto Creek Surface Water AMP, and made minor modifications to AMP responses and groundwater evaluation frequencies.

Table of 2018-01 Revisions

Section Revised	Description of Revision
2.1.3	Revisions include the following changes: • Revision to Specific Performance Thresholds and Responses in Table 2-1. • Revision to Specific Performance Thresholds at W2 in Table 2-2.

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1 Introduction

1.1 Overview

Minto Explorations Ltd. (Minto), a wholly owned subsidiary of Capstone Mining Corp. (Capstone), owns and operates the Minto copper mine. Minto Mine is located within Selkirk First Nation (SFN) Category-A Settlement Land (Parcel R-6A), and is approximately 240 km northwest of Whitehorse, Yukon Territory. The Minto mine commenced commercial operations in October 2007.

The Yukon Government's Decision Document (Yukon Government, 2014) following the YESAA review of Minto's Phase V/VI project proposal (file # 2013-0100, (Minto Explorations Ltd., 2013)) required an Adaptive Management Plan (AMP) for the mine operations. As an outcome of the Phase V/VI mine licensing processes, Minto is required to update the AMP with additional information outlined in its Quartz Mining License QML-0001 (QML) and Water Licence QZ14-031 (WL).

AMPs are tools used to address uncertainty or conditions beyond those anticipated in mining operations. AMPs outline a range of possible but unexpected outcomes and the responses that will be undertaken to curb possible negative impacts associated with these unexpected situations.

Mining activities are highly managed operations, with very prescriptive and detailed management plans required for both operational control and regulatory approval. More mature mines such as Minto have management plans which benefit from the operational experience at the site, and uncertainty in the range of conditions expected is reduced through this operational experience.

Minto has developed a number of operational management plans which describe the management and response actions for expected conditions at the site. These plans currently include:

- Solid Waste Management Plan;
- Environmental Monitoring, Surveillance, Reporting Plan;
- Wildlife Protection Plan;
- Spill Contingency Plan;
- Sediment and Erosion Control Plan;
- Mine Development and Operations Plan;
- Underground Mine Development and Operations Plan;
- Mill Operations Plan;
- Water Management Plan;
- Tailings Management Plan;
- Waste Rock and Overburden Management Plan;
- Emergency Response Plan;
- Heritage Resources Protection Plan; and

- Reclamation and Closure Plan

This AMP is intended to provide a framework for responses to conditions beyond those expected and identified in these decision-based management plans. Consequently, this AMP addresses a limited range of components.

1.2 Adaptive Management Planning

Adaptive management is an approach to environmental management that is appropriate when a mitigation measure may not function as intended or when broad-scale environmental change is possible. Adaptive management plans are precautionary in nature, and provide a level of security in long term environmental planning. Adaptive management plans also allow for the inclusion of improved science into mitigation measures as they are continually revised.

Adaptive management has been evolving since its emergence in the 1970s. Adaptive approaches include an ability to incorporate knowledge into the management plan as the knowledge is gleaned and circumstances change. Eberhard et al. (Eberhard, et al., 2009) described the categories of knowledge that may trigger changes to water quality management plans; system understanding, measuring progress and anticipating changes. These categories allow for the inclusion of knowledge and adaptation of management to changed conditions. Embedding adaptation into environmental plans involves thinking about how the results of monitoring will change management actions. Adaptive management plans are a way to accept uncertainties and build a structured framework to respond to changing conditions.

Adaptive management conducts a flexible path with actions to take when specific triggers occur. AMPs are a formalization of a plan for performance monitoring and project re-evaluation in the future. The general structure of adaptive management can be described by the following steps:

1. Identify risk triggers associated with vulnerabilities or uncertainties;
2. Quantify impacts and uncertainties;
3. Evaluate strategies and define implementation path that allows for multiple options at specific triggers;
4. Monitor the performance and critical variables in the system; and
5. Implement or re-evaluate strategies when triggers are reached.

Although there are no widely used AMP terms, the steps listed above are representative of typical AMP processes. Within AMPs, triggers provide decision points in a stepwise decision-making framework that identifies how and when management action should be taken. A key characteristic of adaptive management is monitoring, which is used to advance scientific understanding and to adjust management policies in an iterative process. Adaptive management is a rigorous method for addressing uncertainties in ecosystem management.

1.3 Adaptive Management Plan Objectives

An AMP is a management tool wherein a framework is provided to make quick and effective decisions to guide responses to unforeseen events. This document identifies areas of uncertainty within the operational phase of the Minto Mine life and provides an AMP framework for each. For each component the AMP describes monitoring commitments, thresholds, triggers and responses to underperforming elements or emerging risks within the

component. The steps laid out in the AMP framework are precautionary, and therefore they provide the confidence that action will be taken before adverse environmental impacts are observed.

Response planning, and results for anticipated events are contained within site management plans while AMPs guide responses to unforeseen or contingency events. This AMP provides a framework to guide responses to unanticipated monitoring results and to potential but low probability events where uncertainty exists.

It is difficult to predict the specific environmental condition that may arise which requires a response from management and, therefore, the AMP does not provide specific detailed descriptions of responses to a situation. The AMP provides a range of possible responses to use as a guide to respond to specific environmental conditions encountered. Management should use the information provided in the AMP and undertake the appropriate response.

1.3.1 Updated AMP Objectives

This version of the AMP has been updated to include the requirements of the QML and WL.

The QML was issued December 18, 2014 and was accompanied by a letter entitled outlining the components required in an Adaptive Management Plan (Yukon Government - Energy, Mines and Resources, 2014). The AMP includes the requirements outlined in the letter with the exception of Cover Systems, as they are more appropriately addressed in the Closure Adaptive Management Plan which forms part of the Reclamation and Closure Plan. Additionally, two conditions have been adopted into this plan as part of the March 2nd, 2016 EMR approval of the AMP.

The WL was issued August 5, 2015 and the requirements for the AMP are outlined in clause 109 (Yukon Water Board, 2015).

Clause 109 details and the sections in the AMP where these are addressed are summarized in Table 1-1.

Table 1-1: QZ14-031 Concordance table

109) The Licensee shall submit to the Board for Review and Approval an updated Operational Adaptive Management Plan. This plan shall be submitted by December 18, 2015, and shall be implemented once approved. The updated plan shall include, without limitation:	Section addressed
a) a surface water quality Adaptive Management Plan for McGinty Creek;	2.2
b) groundwater quality Adaptive Management Plans, including establishment of thresholds, for the Minto Creek and McGinty Creek watersheds;	2.3, 2.4
c) freeboard thresholds for each water storage facility;	2.5
d) WQOs, as stated in clause 8; (note: Clause 8 is presumed to be incorrect reference, and therefore W2 WQOs from licence have been included)	2.1
e) specifications for Monthly Reports that will include, but not be limited to the activities carried out under the Adaptive Management Plan, and	3.1
f) specifications for the Annual Report to include but not be limited to: i. activities undertaken in relation to the Adaptive Management Plan; ii. trend analysis and water levels in Minto and McGinty creeks; iii. proposed updates and revisions to the Adaptive Management Plan, and iv. any other revisions	3.1, 3.2

1.4 Adaptive Management Plan Approach

In addition to the conclusions drawn from research, the approach presented in this AMP follows the Environmental Code of Practice for Metal Mines, Section 4.1.17 on Adaptive Management:

"Mine owners/operators should use adaptive management methods to revise and refine the environmental management strategy. Adaptive management should consider a wide range of factors, including:

- The results of environmental audits or other evaluation activities;
- The results of environmental monitoring;
- The results of monitoring of the performance or condition of environmental infrastructure, such as containment structures, water management systems or treatment facilities;
- Technological developments; and
- Changing environmental conditions." (Environment Canada, 2009)

In addition to the guidance provided by the Environmental Code of Practice for Metal Mines, the AMP serves to meet the Yukon Government's Decision Document following the YESAA review of Minto's Phase V/VI project proposal which identifies some areas that an AMP for operations should be prepared to address including "*water quality, physical stability, covers, water treatment, and water management;".* Though some covers are anticipated to be placed as part of progressive reclamation, they are not an operational feature and therefore have not been included in this AMP.

1.4.1 AMP Components

The following AMP components have been identified as having the potential for unexpected conditions during the operational period for which the Operational Management Plans may not provide adequate mitigation against potential effects to the environment or human health and safety:

- Surface water quality;
- Groundwater quality;
- Water Management, and
- Physical Stability

The specific AMP framework for these components is described in subsequent sections.

1.4.2 AMP Framework

The AMPs for each component are laid out using a common element approach to create consistency in implementation of the AMP protocol for all components as illustrated in Figure 1-1. The common elements are:

1. Description of the component
   - *Description* - description and understanding of the component leads to risk narrative and specific performance thresholds.
   - *Risk Narrative* describe the possible environmental impacts and environmental conditions that implementation of the AMP will prevent.
2. Monitoring the component

- *Specific Indicators* are the environmental or physical parameters to be monitored and assessed. Specific indicators are measurable or observable, and are indicative of changes from the designed or expected condition.
- *Monitoring Requirements* describes the monitoring regime for the component including frequency, type of data required and interpretation of results.
- *Specific Performance Thresholds* define the conditions, in terms of specific indicators, when action is triggered. Performance thresholds are staged to accommodate levels of concern and a diversity of actions. To the extent possible, specific performance thresholds will include early warning thresholds so that timely and informative responses are initiated before higher impact thresholds are triggered. Trend analysis at early warning thresholds are included to determine the potential of triggering subsequent thresholds.

3. Responding to unexpected conditions of the component

   Specific Responses are staged according to specific performance thresholds describes the actions to be implemented if specific performance thresholds are crossed. They are provided in the following categories:

   a) Notification
   b) Review
   c) Evaluation
   d) Action

4. Annual Reporting and Review

   Annual Reporting reflects annual changes made to the AMP as the site conditions change. The AMP should be modified whenever unexpected circumstances are encountered and the protocol is implemented or when additional proven science or technology becomes available. The annual review will include a review of the relevant monitored data and AMP elements. Updates, amendments, performance thresholds crossed, and trigger(s) activated will be provided to the appropriate governmental (including SFN) organizations as required and will be part of the annual report.

   Additional reporting is described further in section 3.1.

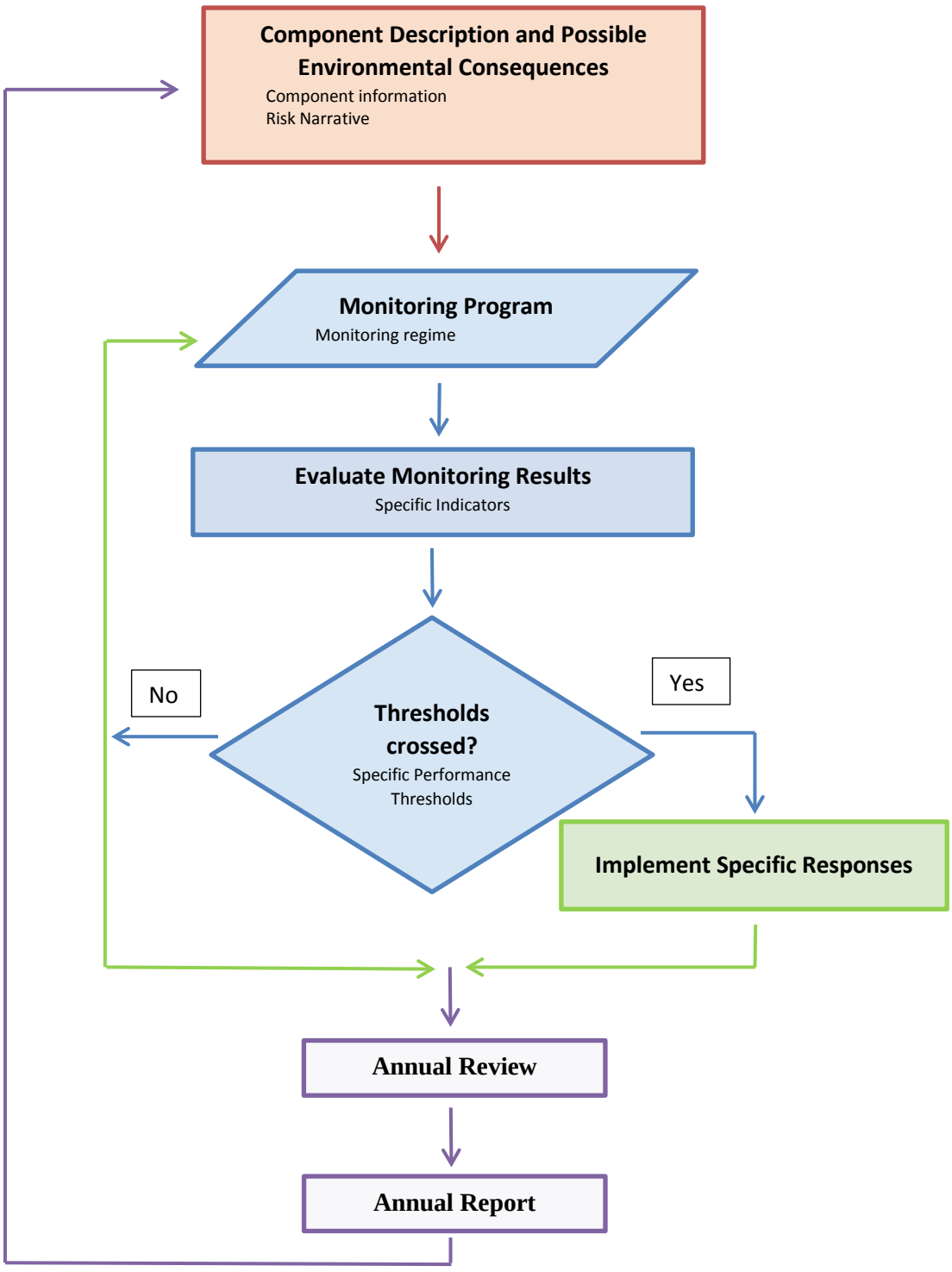


Figure 1-1: Sequential Components of the AMP (Adapted from AECOM 2010)

2 Adaptive Management Plans for Mine Components

2.1 Minto Creek Surface Water Quality

2.1.1 Description

Station W50 is considered the main control point on Minto Creek. It is also the last surface monitoring point on the mine site property and is considered a discharge compliance point under the current water licence. Additional key surface water monitoring locations upgradient of W50 and within the mine footprint are located at W16, W17, W15, W35, and W37.

As described in the Water Management Plan (Minto Explorations Ltd., 2015), Minto maintains substantial flexibility over the control and management of site runoff. The conveyance, storage and treatment systems are oriented to adequately manage site water to meet the current and proposed discharge standards at W50.

The W2 monitoring point near the Yukon River is beyond the final mine water discharge point controlled by Minto, and is subject to a broad range of influences from tributaries and catchment areas that are beyond Minto's control. However, Minto is committed to monitoring the water quality at W2 and responding to changing water quality in lower Minto Creek as appropriate. The AMP framework below compliments the operational water management plan at the mine site with a decision-based structure for ensuring that negative impacts to lower Minto Creek from mining activities are avoided.

2.1.2 Risk Narrative

Increase in contaminant concentrations from the mine causes adverse effects to aquatic resources in the receiving environment (lower Minto Creek) despite adherence to discharge standards.

2.1.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to water quality and the monitoring program are provided below in Table 2-1. Specific thresholds are identified in Table 2-2. Specific thresholds are based on 70 and 85 percent of the WQO for each parameter, as indicators and triggers for action before WQOs are reached.

The monitoring results that will be evaluated and utilized in this component of the AMP are a requirement of the Surface Water Surveillance Program of the Environmental Monitoring, Surveillance and Reporting Plan (EMSRP) (Minto Explorations Ltd., 2016). The monitoring data will be compared to the specific performance thresholds monthly (by the end of the month following the month in which samples were collected) - this corresponds to the existing monthly reporting schedule.

Table 2-1: Specific Indicators, Performance Thresholds and Responses for Surface Water Quality at Station W2 in lower Minto Creek.

Specific Indicators	Specific Performance Thresholds	Specific Responses
Aqueous Concentrations at Station W2 for the following parameters with water quality objectives Parameters: • NH_4-N • NO_2-N • NO_3-N • pH • Quarterly Bioassay Dissolved • Aluminum • Arsenic • Cadmium • Chromium • Copper • Iron • Lead • Molybdenum • Nickel • Silver • Selenium • Zinc (See Table **2-2** below for specific threshold values)	**Specific Threshold 1** • Exceedance of ST1 value in two consecutive samples **(scheduled or re-sample)**	**Notification** • Minto Management • Include in scheduled Water Use Licence monthly reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • Compare with W3 results • A trend analysis will be conducted by Minto's senior level environmental personnel. **Action** • If comparison with W3 result indicates mine loadings are responsible for exceedance then: o Expedite results for subsequent sampling and review the data to see if the exceedance continues or if no other sample has been collected then re-sample within 24-hour of original sample result review. • Actions will continue until performance thresholds are no longer exceeded.
	Specific Threshold 2 • Exceedance of ST2 value in two consecutive samples **(scheduled or re-sample) at W2** where evaluation confirmed mine loading responsible for first exceedance	**Notification** • Minto Management • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • Compare with W3 results. If comparison with W3 result indicates that mine loadings are responsible for exceedance; and validation confirms original result, then: o Evaluate causes for load contributions and develop investigation plan. • A trend analysis will be conducted by Minto's senior level environmental personnel. **Action** • Implement investigation plan, including at a minimum: o Expedite results for subsequent sampling and review the data to see if the exceedance continues or if no other sample has been collected then re-sample within 24-hour of original sample result review; and o Site investigation of candidate load contributions. • Review results of investigation and prepare recommendations if appropriate. • Implement recommendations. • Actions will continue until performance thresholds are no longer exceeded. • If trend analysis suggests WQO exceedance within one year, then initiate actions for Specific Threshold 3.
	Specific Threshold 3 • WQO exceeded at W2 in a single sample	**Notification** • Minto Management, SFN, YG Inspector • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol. o Review laboratory QA/QC report. o Verify original result, or re-run sample if laboratory error indicated. **Evaluation** • Compare with W3 results. If comparison with W3 result indicates mine loadings responsible for exceedance – and verification confirms original result – then: o Evaluate candidate causes for load contributions and develop investigation plan (or review/revise as appropriate).

Specific Indicators	Specific Performance Thresholds	Specific Responses
		• A trend analysis will be conducted by Minto's senior level environmental personnel. **Action** • Maintain weekly monitoring at W2 and W3. • Implement investigation plan, including any reviews/revisions, and at a minimum: o Expedite results for subsequent sampling and review the data to see if the exceedance continues or if no other sample has been collected then re-sample within 24-hour of original sample result review. o Site investigation of candidate load contributions; and o Engage a qualified individual for the evaluation of potential effects to aquatic resources (i.e. Compare to Predicted No Effect Concentration (PNEC) and apply Biotic Ligand Model (BLM) for D-Cu, invertebrate tissue for Se, compare with calculated acute guideline for D-Cd.) This may include but not be limited to the evaluation of existing data and results from effluent toxicity, surface water toxicity, sediment chemistry, sediment toxicity, benthic invertebrate community structure and fish health testing/programs. • Review results of investigation and prepare recommendations if appropriate. • Implement recommendations. • Actions will continue until performance thresholds are no longer exceeded.
	Specific Threshold 4 • WQO or PNEC exceeded in **2 consecutive samples (scheduled or re-sample) at W2** where evaluation confirmed mine loading responsible for first exceedance	**Notification** • Minto Management, SFN, YG Inspector • Include in scheduled Water Use Licence reporting **Review** • Compare with W3 results • Follow QA/QC investigative protocol **Evaluation** • Compare with W3 results. If comparison with W3 result indicates mine loadings responsible for exceedance; and verification confirms original result, then: o Evaluate candidate causes for load contributions and develop investigation plan (or review/revise as appropriate). o Provide investigation plan to SFN/YG Inspector. • Evaluate potential for mine loadings to cause adverse effects to aquatic resources (i.e. redo BLM modeling). • Consider ongoing WQ monitoring results in development of investigation plan. • A trend analysis will be conducted by Minto's senior level environmental personnel. **Action** • Maintain weekly monitoring and collect samples at greater frequency as required. • Develop investigation plan, including at a minimum: o Expedite results for subsequent sampling and review the data to see if the exceedance continues or if no other sample has been collected then re-sample within 24-hour of original sample result review. o Site investigation of load contributions; and o Evaluation of potential effects to aquatic resources. • Review results of investigation and prepare recommendations. • Implement recommendations arising from investigations. • Implement necessary reasonable and practical measures to reduce contaminant loading from mine to Minto Creek. • Suspend discharge from the mine until water quality is appropriate for discharge. • Actions will continue until performance thresholds are no longer exceeded. **If threshold consistently exceeded for 2 months**, then: • Develop revised forecast for near-term (12 months) water quality in Minto Creek. • Develop and implement any additional mitigation measures to reduce loading from mine site, if necessary, with appropriate regulatory approvals. • Actions will continue until performance thresholds are no longer exceeded.

Table 2-2: Specific Performance Thresholds for Surface Water Quality in Lower Minto Creek (W2)

	Specific Threshold 1 (ST1) (70% of WQO)	Specific Threshold 2 (ST2) (85% of WQO)	W2 Water Quality Objective (WQO)
	mg/L	mg/L	(mg/L)
Ammonia - N, mg/L	0.175	0.2125	0.25
Nitrite - N, mg/L	0.042	0.051	0.06
Nitrate - N, mg/L	6.37	7.735	9.1
Aluminum (dissolved), mg/L	0.07	0.085	0.1
Arsenic (dissolved), mg/L	0.0035	0.00425	0.005
Cadmium (dissolved), µg/L	$0.7*e^{(0.736(\ln(\text{hardness})-4.943)}$	$0.85*e^{(0.736(\ln(\text{hardness})-4.943)}$	$e^{(0.736(\ln(\text{hardness})-4.943)}$
Chromium (dissolved), mg/L	0.0007	0.00085	0.001
Copper (dissolved), mg/L (when [DOC] @ W2 >10 mg/L)	0.014	0.017	0.02
Copper (dissolved), mg/L (when [DOC] @ W2 ≤10 mg/L)	0.0091	0.01105	0.013
Iron (dissolved), mg/L	0.77	0.935	1.1
Lead (dissolved), mg/L	0.0028	0.0034	0.004
Molybdenum (dissolved), mg/L	0.0511	0.06205	0.073
Nickel (dissolved), mg/L	0.077	0.0935	0.11
Silver (dissolved), mg/L	0.00007	0.000085	0.0001
Selenium (dissolved), mg/L	0.0014	0.0017	0.002
Zinc (dissolved), mg/L	0.021	0.0255	0.03
pH (pH units)	n/a	n/a	6.0 – 9.0

Water Use Licence QZ14-031 identifies the WQO Station as:

- Station W2 during the period when flow is encountered at stations W15 and W35; or
- Station W50 during the period when flow is not encountered at stations W35 and W15.

Clause 11 of the licence identifies the water quality objectives (above in Table 2-2) for Minto Creek, and states that “any exceedances of these at the defined WQO Station shall trigger the Operations Adaptive Management Plan.” The AMP framework in this section ensures that action under the Operational AMP is triggered in advance of exceedances of WQOs at station W2. In the event of the WQO station being W50 (no flow at stations W35 and W15), it is assumed that this would be under winter conditions, and that there will be no contributing flow from the rest of the Minto Creek catchment downstream of the mine. In this case, Minto would adhere to the simple AMP outline listed below:

Table 2-3: Specific Indicators, Performance Thresholds and Responses for Surface Water Quality at Station W50 in Minto Creek.

Specific Indicators	Specific Performance Thresholds	Specific Responses
Aqueous Concentrations at Station W50 for the following parameters with water quality objectives Parameters: • NH_4-N • NO_2-N • NO_3-N Dissolved • Aluminum • Arsenic • Cadmium • Chromium • Copper • Iron • Lead • Molybdenum • Nickel • Silver • Selenium • Zinc (See Table **2-2** above for specific threshold values)	**Specific Threshold** • Exceedance of WQO for in a single sample	**Notification** • Minto Management • Include in scheduled Water Use Licence reporting **Action** • Suspend discharge from the mine until water quality is appropriate for discharge.

Minto will evaluate and respond to toxicity testing in mine effluent as outlined in Table 2-4 below:

Table 2-4: Specific Indicators, Performance Thresholds and Responses for Toxicity Testing at Stations W3 and W50 in Minto Creek.

Specific Indicators	Specific Performance Thresholds	Specific Responses
• Licensed Bioassay test and analysis (acute or chronic) at W3 or W50	**Specific Threshold** • Failure of any licensed effluent bioassay test	**Notification** • Minto Management • Include in scheduled Water Use Licence reporting **Action** • If not already initiated, collect Bioassay samples at W2: 7-day rainbow trout test (Lazorchak and Smith 2007), 7-day for *Ceridaphnia dubia* (EPS 1/RM/21), and continue quarterly until directed by a qualified professional to stop.

2.2 McGinty Creek Surface Water Quality

2.2.1 Description

The Minto North deposit is an extension of the mineralized corridor being mined within the Minto Creek catchment. It is within the McGinty Creek catchment area, to the north of Minto Creek. Minto has been monitoring surface water quality in the McGinty Creek catchment since 2009, and the results from monitoring program (until the end of 2015) are presented in McGinty Creek Water Quality Characterization, July 2016 (AEG, 2016).

As described in the Water Management Plan (Minto Explorations Ltd., 2015), during the recently completed active mining of Minto North, Minto actively managed water directly impacted by open pit mining with pump trucks. With surface mining in Minto North now complete, there is no active water management, with runoff and meteoric water now allowed to accumulate in the pit.

The AMP framework below provides a decision-based structure with the goal of avoiding changes to background water quality in lower McGinty Creek that result from completed mining activities at Minto North. Station MN-4.5 is the monitoring station on the lower main stem of McGinty Creek, after the north and south tributaries converge, and near the confluence of McGinty Creek with the Yukon River. Surface water quality at the MN-4.5 monitoring point is subject to periodic TSS influences from catchment area that are beyond Minto's control, so dissolved metals concentrations will be used to track influences from the Minto North development.

2.2.2 Risk Narrative

Increase in contaminant concentrations from completed Minto North mining activities causes unacceptable changes to surface water quality in McGinty Creek.

2.2.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to water quality and the monitoring program are provided below in Table 2-6. The specific indicators are total suspended solids and contaminant (nitrogen species and dissolved metals/metalloids) concentrations for parameters identified in the Water Use Licence QZ14-031 Table 2 – Water Quality Objectives. Thresholds are all based on a proposed water quality objective (WQO), and the specific indicators station, as identified above, is MN-4.5. The discussion below applies to data collected from this station. The selection of this indicator station, along with other aspects of the AMP framework, is consistent with the approach taken for closure water quality objective development in Minto Creek (i.e. downstream indicator station, non-degradation water quality objectives, and a focus on dissolved metal concentrations.)

Essentially, Minto has adopted the statistical definitions of non-degradation (from the discussions and agreement with SFN regarding Minto Creek closure water quality objectives) as the basis of developing these revised McGinty Creek Water Quality Objectives. Monthly monitoring has continued in the McGinty Creek catchment, and the background dataset has been updated to include all monitoring data from initiation of the program in May 2009 until July 2015 (stripping of the Minto North Pit began in August 2015.) Similar to the data treatment used in Minto Creek, data from monitoring stations were collapsed into monthly results (most monitoring has been undertaken monthly anyway, and this was only required for May 2009 when sampling was weekly.) All monthly data were then used to calculate the 95th percentile, for use as the maximum water quality objective, or individual data point evaluator (IDPE). The monthly data were then also grouped by year, and the annual medians were calculated. The 95th percentile of these annual medians was calculated for each station to generate the central tendency evaluator (CTE).

These objectives form the basis of the AMP thresholds. Values lower than the WQOs have been selected as early warning thresholds (ST 1) prior to the WQOs themselves forming the higher level specific thresholds (ST2 and ST3). For these thresholds, the 85th percentile of the same data selected. Utilizing a statistic such as this is more effective and reliable than using a percentage (e.g. 75%) of maximum value, as it considers the actual statistical distribution of the background data.

The actual calculated threshold values are presented in Table 2-5.

The monitoring results that are evaluated and utilized for this component of the AMP are a requirement of the Surface Water Surveillance Program of the EMSRP (Minto Explorations Ltd., 2016). The monitoring data will be compared to the specific performance thresholds monthly (by the end of the month following the month in which samples were collected) - this corresponds to the existing monthly reporting schedule.

Table 2-5: Specific Thresholds for McGinty Creek, Station MN-4.5

Analytes	Specific Threshold 1 (85th percentile)	Water Quality Objectives (Specific Thresholds 2 and 3)	
		Individual Data Point Evaluator (95^{th} percentile)	Central Tendency Evaluator (95th percentile of annual medians)
Total Suspended Solids (mg/L)	52.0	269	32.0
Ammonia (mg/L)	0.040	0.12	0.046
Nitrite (mg/L)	0.006	0.05	0.005
Nitrate (mg/L)	0.200	0.232	0.083
Dissolved Aluminum (µg/L)	48.6	135.0	47.0
Dissolved Arsenic (µg/L)	0.55	0.61	0.54
Dissolved Cadmium (µg/L)	0.026	0.041	0.015
Dissolved Chromium (µg/L)	1.0	1.0	1.0
Dissolved Copper (µg/L)	2.9	3.5	2.8
Dissolved Iron (µg/L)	334	403	358
Dissolved Lead (µg/L)	0.20	0.20	0.20
Dissolved Molybdenum (µg/L)	1.0	1.0	1.0
Dissolved Nickel (µg/L)	1.6	1.8	1.6
Dissolved Selenium (µg/L)	0.17	0.20	0.16
Dissolved Silver (µg/L)	0.020	0.020	0.020
Dissolved Zinc (µg/L)	5.0	5.2	5.0

Table 2-6: Specific Indicators, Performance Thresholds and Responses for Surface Water Quality in McGinty Creek

Specific Indicators	Specific Performance Thresholds	Specific Responses
Aqueous Concentrations at Station MN-4.5 for parameters with Water Quality Objectives. (See Table 2-5 above for Water Quality Objective and threshold values.)	**Specific Threshold 1** • Exceedance of ST1 value in two consecutive samples **(scheduled or re-sample)**	**Notification** • Minto Management • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • Compare with MN-1.5 results • A trend analysis will be conducted by Minto's senior level environmental personnel. **Action** • If comparison with other results suggests that Minto North loading may be responsible for exceedance and validation confirms original result, then: o Evaluate causes for load contributions, and o If trend analysis suggests WQO exceedance within one year, then initiate actions for threshold 2.
	Specific Threshold 2 • Exceedance of the WQO (grab vs. IDPE, or 12-month moving average vs. CTE)	**Notification** • Minto Management • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • Measures from ST1, and • Engage a qualified professional to evaluate potential effects to aquatic resources • A trend analysis will be conducted by Minto's senior level environmental personnel. **Action** • If comparison with results suggests that Minto North loadings are responsible for exceedance and validation confirms original result, then: o Re-sample MN-1.5 and MN-4.5 within **two weeks** of original sample result review; and o Evaluate causes for load contributions o Develop a mitigation strategy with recommendations based on the findings of the potential effects to aquatic resources evaluation • Actions will continue until performance thresholds are no longer exceeded.
	Specific Threshold 3 • Exceedance of the WQO (as defined above) in **2 consecutive samples (scheduled or re-sample)** where evaluation confirmed mine loading responsible for first exceedance	**Notification** • Notify management, SFN and YG • include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • Measures from ST2 **Action** • If comparison with results (evaluation, above) suggests that Minto North loadings are STILL responsible for exceedance; and validation confirms original result, then: o If not already implemented, increase monitoring frequency. o Implement recommendations from mitigation strategy (threshold 2 response). This could include batch water treatment in the Minto North pit if determined feasible and appropriate. o Actions will continue until performance thresholds are no longer exceeded.

2.3 Groundwater Quality in Minto Creek Watershed

2.3.1 Description

Groundwater quality has the potential to be important in terms of contributions to surface water quality. Groundwater contributes to streamflow as baseflow, which is typically most important during the autumn/winter low flow season when surface water flows are minimal.

All mine workings and waste facilities within in the Minto Creek catchment are located upgradient (and west) of the Water Storage Pond (WSP). The monitoring supporting the AMP framework is defined and described in the Groundwater Monitoring Plan (GMP) (Minto Explorations Ltd., 2016). Groundwater is monitored both upgradient (west) and downgradient (east) of the WSP; upgradient monitoring is carried out via a multi-level monitoring well at MW12-06, and downgradient monitoring is carried out via a multi-level monitoring well at MW12-05. Surface water downgradient of the WSP is monitored at several stations including station W3, which is located immediately adjacent to MW12-05.

As described in the 2015 Groundwater Model Update (SRK 2015a), groundwater coming from the mine area (the western and highest elevation portion of the Minto Creek catchment) is expected to discharge to surface water in the vicinity of the Water Storage Pond (i.e. upgradient of monitoring well MW12-05). Minimal groundwater from the mine area is expected to discharge to Minto Creek down gradient of the Water Storage Pond. MW12-05 and MW12-06 are optimally located to monitor expected groundwater flow paths.

The W2 surface water monitoring station is located at approximately 600 m from the Yukon River and is 6 km downstream of the Minto lease boundary. Water reporting to W2 is subject to influences from groundwater and surface water outside the mine area. Minto is committed to monitoring groundwater quality at MW12-05 and MW12-06 and surface water quality at W2 as required by Water Use Licence QZ14-031, and responding to changing water quality in groundwater and lower Minto Creek as appropriate.

The operational AMP framework below complements the operational water management plan at the mine site with a decision-based structure for ensuring that negative impacts to lower Minto Creek from mining activities are avoided.

2.3.2 Risk Narrative

Flux of geochemical load from the mine via groundwater pathways causes surface water quality objectives to be in exceeded in Minto Creek at station W2.

2.3.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to groundwater quality in Minto Creek watershed are provided in Table 2-7. Specific Performance Thresholds (SPTs) are defined for each of the Effluent Quality Standards (EQS) parameters identified in Clause 9 (a), Table 1 of Water Use Licence QZ14-031, with the exception of pH, oil and grease, iron and nitrite (as discussed later in this section). Additionally, although it is not specified in Clause 9(a), sulphate has been included in the SPTs. Table 2-8 compiles the concentration of background groundwater for each Specific Indicator and the Specific Performance Threshold values.

The concentrations of the background groundwater are based on the respective median parameter concentrations at monitoring well MW09-03 (including all monitoring ports). At present (January 2017), the MW09-03 monitoring record is considered to be the most representative indicator of the baseline groundwater

conditions for the entire project site. MW09-03 was installed in 2009 and monitored to collect baseline data downgradient of the Minto North ore body- this well is located near the southern limit of the McGinty Creek catchment close to the surface water divide with Minto Creek. Pre-mining groundwater concentrations for certain parameters such as cadmium and iron have been relatively high, likely due to the adjacent highly mineralized zone. Mining of this ore body began in Q3 2015 and as such the monitoring record prior for the 2009-2015 period reflects baseline conditions. Another groundwater well (nominally MW16-08, likely MW2017-08 when completed) will be installed in 2017 to allow groundwater quality monitoring within the Minto Creek catchment upgradient of any mine disturbances. Once an adequate record of the groundwater quality from MW16-08 becomes available, the groundwater quality at MW09-03 and MW16-08 will be compared by a qualified professional to verify and assess if background groundwater concentrations used in the current operational AMP remain appropriate. Minto expects that an adequate record will be available after 3 years of monitoring. Other new wells that are planned for installation in 2017 will be monitored in accordance with Minto's EMSRP, but will not be included in the AMP.

Iron and nitrite have been excluded from the Specific Indicators for groundwater quality because of the magnitude of the natural variability observed in groundwater at Minto. In addition, pH as also been excluded as a Specific Indicator for groundwater quality because it is not as useful an early warning indicator as sulphate and metal/ metalloid concentrations, and Oil & Grease has been excluded because it is not relevant as a specific indicator for groundwater. Although iron, nitrite and pH are excluded from the Specific Indicator list, these three constituents are monitored and would be included in the review of groundwater quality if an SPT was exceeded.

Three SPTs have been defined for the Minto Creek watershed. Rationale for development of the SPTs is as follows:

- SPT-1 corresponds to a trend-based assessment designed to flag a potential rapid increase in groundwater loadings that has not yet exceeded concentration-based thresholds. The assessment is structured to determine if an indicator has increased significantly compared to the last sampling event. The assessment will be performed as followed:

  $$((C_n - C_{n-1})/(C_{SPT-2} – C_n) > 0.2$$

  Where:

  - C_n = the parameter concentration of groundwater from the latest sampling event;
  - C_{n-1} = the parameter concentration of groundwater from the last sampling event ;
  - C_{SPT-2} = the parameter concentration for the SPT-2;

  The SPT-1 provides a conservative threshold considering that it is weighted against the concentration for the SPT-2. The SPT-1 will be increasingly sensitive to change in concentrations between two sampling events as groundwater approaches the SPT-2, since the size of the denominator decreases as the SPT-2 value is approached.

- SPT-2 generally corresponds to the EQS concentrations defined in Clause 9 (a), Table 1 of Water Use License QZ14-031 (with four exceptions (Cr-D, Cu-D, Ni-D and sulphate) as indicated in the notes to Table 2-7). The EQS are defined for surface water discharge, and the mine is not permitted to discharge water that exceeds any EQS guidelines to surface water. As SPT-2 applies to groundwater (not surface water) concentrations in single zone, this is a highly conservative threshold.

- SPT-3 generally corresponds to the estimated concentrations in groundwater that would be necessary to cause exceedance of the Water Quality Objectives in lower Minto Creek, at W2, under long term steady state conditions (exceptions are listed in the notes to Table 2-7). These concentrations were determined by conducting a mass loading calculation to determine the groundwater concentrations that would be necessary to cause exceedance of WQOs in lower Minto Creek during low flow periods where all streamflow is derived from groundwater discharge. The mass loading calculation was structured to represent a low-flow period when all surface flows in lower Minto Creek originate from groundwater discharge to the creek. The calculation was done using the following formula:

$$C_{gw\text{-}mine_max} = ((Q_{W2} \times C_{W2\text{-}WQO}) - (Q_{gw\text{-}bgrnd} \times C_{gw\text{-}bgrnd}))/Q_{gw\text{-}mine}$$

Where:

- $C_{gw\text{-}mine_max}$ = the indicator parameter concentration of all groundwater upgradient of MW12-05 that reports to Minto Creek that would be required to cause surface water to exceed the WQO at station W2.
- Q_{W2} = the combined groundwater flow discharging to Minto Creek (total groundwater discharge to Minto Creek from the 2015 groundwater model update) during low flow periods
- $C_{W2\text{-}WQO}$ = the WQO parameter concentration for surface water at W2
- $Q_{gw\text{-}bgrnd}$ = background groundwater flow discharging to Minto Creek (groundwater discharge down gradient of the Water Storage Pond from the 2015 groundwater model update) during low flow periods
- $C_{gw\text{-}bgrnd}$ = background groundwater concentration (based on median concentrations from the 2009-2015 baseline monitoring period in MW09-03)
- $Q_{gw\text{-}mine}$ = the estimated groundwater flow from the mine at the Water Storage Pond (from the 2015 groundwater model update)

If groundwater concentration of one indicator reached the SPT-3 in an individual port, the quality of the lower Minto Creek at W2 would not yet exceed the Water Quality Objective due to the contribution of groundwater from un-impacted areas of the Minto Creek watershed and the fact that groundwater concentrations at a specific monitoring well port represents only a portion of the flow and not the whole groundwater flow field. The SPT-3 provides therefore a conservative threshold for action before any significant effect would be observed in surface water.

The monitoring results that are evaluated and utilized for this component of the AMP are a requirement of the Groundwater Monitoring Program of the EMSRP (Minto Explorations Ltd., 2016). The monitoring data will be compared to the specific performance thresholds monthly (by the end of the month following the month in which samples were collected) - this corresponds to the existing monthly reporting schedule. The evaluation of new AMP triggers will be done by the qualified professional (QP) with the monthly data review as indicated in the response Table 2-7. If however, the exceedance is an on-going trigger, then the review by the QP will be quarterly for that parameter in the specific zone.

Examples of actions that may arise from recommendations include:

- Continuation of monitoring;

- Continuation monitoring with an increase in monitoring frequency;
- Development of additional monitoring points and monitoring of those newly-established monitoring locations;
- Completion of appropriate risk assessment;
- Development and execution of a focused study to better understand the cause of exceedance.

Table 2-7: Specific Indicators, Performance Thresholds and Responses for Groundwater Quality in Minto Creek

Specific Indicators	Specific Performance Thresholds	Specific Responses
Aqueous concentrations in samples collected from multi-level groundwater monitoring wells MW12-05 or MW12-06 for the following parameters with water quality objectives: • Dissolved Aluminum • Dissolved Arsenic • Dissolved Cadmium • Dissolved Chromium • Dissolved Copper • Dissolved Iron • Dissolved Lead • Dissolved Molybdenum • Dissolved Nickel • Dissolved Silver • Dissolved Selenium • Dissolved Zinc • NH_4-N • NO_3-N • Sulphate	**SPT-1** • Exceedance of SPT-1 for any sample collected during routine monitoring from multilevel groundwater monitoring wells MW12-05 or MW12-06.	**Notification** • Minto Management • Include in scheduled Water Use Licence monthly reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated o Timing: initiate within 1 week of triggering SPT **Evaluation** • Review of groundwater monitoring data (trend analysis included) to be undertaken by qualified professional, and appropriate recommendations to be developed o Review must consider the risk narrative (i.e. exceedance of surface water quality objectives as a result of groundwater flux) • Submit recommendations to regulator for review and approval o Timing: submit within 1 week of receipt of recommendations **Action** • Follow recommendations arising from review undertaken by qualified professional. Include the trend analysis in scheduled reporting o Timing: initiate implementation of recommendations within 1 month of receipt of approval from regulator
	SPT-2 • Exceedance of SPT-2 concentrations in 2 consecutive samples (scheduled or re-sampled) collected during routine monitoring from multilevel groundwater monitoring wells MW12-05 or MW12-06.	**Notification** • Minto Management • Include in scheduled Water Use Licence monthly reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated o Timing: initiate within 1 week of triggering SPT **Evaluation** • Review of groundwater monitoring data (including trend analysis) to be undertaken by qualified professional, and appropriate recommendations to be developed o Review must consider the risk narrative (i.e. exceedance of surface water quality objectives as a result of groundwater flux) o Timing: initiate within 1 week of QA/QC review validating original results • Submit recommendations to regulator for review and approval o Timing: submit within 1 week of receipt of recommendations **Action** • Follow recommendations arising from review undertaken by qualified professional. o Timing: initiate implementation of recommendations within 1 month of receipt of approval from regulator

Specific Indicators	Specific Performance Thresholds	Specific Responses
	SPT-3 • Exceedance of SPT-3 in 2 consecutive samples (Scheduled or re-sampled) collected during routine monitoring from multilevel groundwater monitoring wells MW12-05 or MW12-06.	**Notification** • Minto Management • Include in scheduled Water Use Licence monthly reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated o Timing: initiate within 1 week of triggering SPT **Evaluation** • Review of groundwater monitoring data (including a trend analysis) to be undertaken by qualified professional, and appropriate recommendations to be developed o Review must consider the risk narrative (i.e. exceedance of surface water quality objectives as a result of groundwater flux) o Timing: initiate within 1 week of QA/QC review validating original results • Submit recommendations to regulator for review and approval o Timing: submit within 1 week of receipt of recommendations **Action** • Follow recommendations arising from review undertaken by qualified professional. o Timing: initiate implementation of recommendations within 1 month of receipt of approval from regulator • Increase monitoring frequency to monthly sampling of all monitored zones in the affected multi-level well for a period to be defined by qualified professional.

Table 2-8: Background concentrations and SPTs for the groundwater monitoring in the Minto Creek watershed

Minto Creek	*Concentrations (mg/L)*													
	Ag-D	Al-D	As-D	Cd-D	Cr-D	Cu-D	Pb-D	Mo-D	Ni-D	Se-D	Zn-D	N-NO3	Ammonia	Sulphate[2]
SPT-3 [1]	0.00046	0.48	0.025	0.0060	0.0030	0.060	0.020	0.34	0.33	0.0093	0.13	45	1.0	4951
SPT-2	0.00030	0.30	0.015	0.0030	0.0015[4]	0.030 [4]	0.012	0.22	0.165[4]	0.0060	0.090	27	0.75	1000
Background Groundwater[3]	0.00001	0.0045	0.0001	0.00002	0.00050	0.0014	0.0001	0.006	0.001	0.0002	0.006	0.07	0.051	12

Notes:

1: For most Specific Indicators, SPT-3 is the calculated concentration that all groundwater from the mine catchment must attain to reach the Operational Water Quality Objectives at W2. Model flows were based on 2015 Groundwater Model Update. 15 L/s is the total groundwater discharging to Minto Creek. 3 L/s is the estimated groundwater flow from the Minto Creek catchment up gradient of the Water Storage Pond. The exceptions are dissolved chromium, dissolved copper and dissolved nickel, for which SPT-3 is equal to the Effluent Quality Standard value that applies to surface water discharge from the mine site.

2: There is no Effluent Quality Standard for sulphate. For the SPT-2, the guideline for Aquatic Life from the Contaminated Site Regulation Schedule 3 was used as a replacement.

3: The background concentration in groundwater is calculated as the median of concentrations observed at groundwater monitoring well MW09-03.

4: SPT-2 for dissolved copper, dissolved chromium and dissolved nickel set at one-half (50%) of Effluent Quality Standard (SPT-2 for all remaining parameters (other than sulphate) is equal to (100% of) the Effluent Quality Standard).

2.4 Groundwater Quality in McGinty Creek Watershed

2.4.1 Description

The mine workings in the McGinty Creek catchment are limited to the Minto North Pit- the pit and the overall catchment are described in Section 2.2.1. The catchment area of the Minto North Pit is roughly 15 ha and the catchment area of the McGinty Creek watershed is roughly 3400 ha (SRK 2013); in other words, the Minto North Pit catchment is roughly 0.4% of the total McGinty Creek catchment area. While a groundwater model encompassing the full extent of the McGinty Creek catchment has not been developed, it is clear that only a very small proportion of the groundwater in the McGinty Creek watershed can be affected by the Minto North Pit.

The monitoring supporting the AMP framework is defined and described in the Groundwater Monitoring Plan (part of the EMSRP ((Minto Explorations Ltd., 2016)). Groundwater in the McGinty Creek catchment downgradient of the Minto North Pit is monitored at multi-level monitoring well MW09-03.

2.4.2 Risk Narrative

Flux of geochemical load from the Minto North Pit via groundwater pathways causes surface water quality objectives to be in exceeded in McGinty Creek at station MN4.5.

2.4.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to groundwater quality in McGinty Creek watershed are provided below in Table 2-9. SPTs are defined for each of the EQS parameters identified in Clause 9 (a), Table 1 of Water Use License QZ14-031, with the exception of pH, Oil & Grease, iron and nitrite (for reasons described in the Minto Creek groundwater section (Section 2.3). Additionally, although it is not specified in Clause 9(a), sulphate has been included in the SPTs. The concentration of background groundwater and the Specific Performance Threshold values for each Specific Indicator and each zone (i.e. individual well port) of MW09-03 are compiled in Table 2-10.

As stated above, at present, MW09-03 is considered to be the most representative of the baseline groundwater conditions for the entire project site. MW09-03 was installed in 2009 and monitored to collect baseline data downgradient of the Minto North ore body- this well is located near the southern limit of the McGinty Creek catchment close to the surface water divide with Minto Creek. Pre-mining groundwater concentrations for certain parameters such as cadmium and iron have been relatively high, likely due to the adjacent highly mineralized zone. Mining of this ore body began in Q3 2015 and as such the prior monitoring record from the 2009-2015 period reflects baseline conditions.

Two Specific Performance Thresholds have been defined for McGinty Creek watershed groundwater- both are conservative given the application of the thresholds at individual ports, the small proportion of McGinty Creek catchment groundwater that will be influenced by the mine workings and the expected slow rates of groundwater movement. Rationale for development of the two specific performance thresholds is as follows:

- SPT-1: Three consecutive exceedances of the 75th percentile background level in a single monitoring port.
    - The specification of three consecutive exceedances is intended to avoid triggering the AMP unnecessarily, but to ensure that any sustained increase from baseline conditions receives appropriate scrutiny.
- SPT-2: Three consecutive exceedances of the 95th percentile background level in a single monitoring port.

- The specification of three consecutive exceedances is intended to avoid triggering the AMP unnecessarily, but to ensure that any sustained increase from baseline conditions receives appropriate scrutiny.

The monitoring results that are evaluated for this component of the AMP are an obligation of the Groundwater Monitoring Program of the EMSRP. The monitoring data will be compared to the specific performance thresholds monthly (by the end of the month following the month in which samples were collected) - this corresponds with the existing monthly reporting schedule. The evaluation of new AMP triggers will be done by the qualified professional (QP) with the monthly data review as indicated in the response Table 2-9. If however, the exceedance is an on-going trigger, then the review by the QP will be quarterly for that parameter in the specific zone.

Examples of actions that may arise from recommendations include:

- Continuation of monitoring;
- Continuation monitoring with an increase in monitoring frequency;
- Development of additional monitoring points and monitoring of those newly-established monitoring locations;
- Completion of appropriate risk assessment;
- Development and execution of a focused study to better understand the cause of exceedance.

Table 2-9: Specific Indicators, Performance Thresholds and Responses for Groundwater Quality in McGinty Creek

Specific Indicators	Specific Performance Thresholds	Specific Responses
Aqueous concentrations in samples collected from multi-level groundwater monitoring well MW09-03 for the following parameters: • Dissolved Aluminum • Dissolved Arsenic • Dissolved Cadmium • Dissolved Chromium • Dissolved Copper • Dissolved Lead • Dissolved Molybdenum • Dissolved Nickel • Dissolved Silver • Dissolved Selenium • Dissolved Zinc • NH_4-N • NO_3-N • Sulphate	**Specific Threshold 1** • Three consecutive exceedances of the 75th percentile value* from the baseline period (2009-2015) in routine monitoring results from a single monitoring port in MW09-03. *values provided in Table 2-9	**Notification** • Minto Management • Include in scheduled Water Use Licence monthly reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original results, or re-run samples if a laboratory error is indicated o Timing: initiate within 1 week of triggering SPT **Evaluation** • Review of groundwater monitoring data (including trend analysis) to be undertaken by qualified professional, and appropriate recommendations to be developed o Review must consider the risk narrative (i.e. exceedance of surface water quality objectives as a result of groundwater flux) o Timing: initiate within 1 week of QA/QC review validating original results • Submit recommendations to regulator for review and approval o Timing: submit within 1 week of receipt of recommendations **Action** • Follow recommendations arising from review undertaken by qualified professional. o Timing: initiate implementation of recommendations within 1 month of receipt of approval from regulator • Trend analysis to be included in scheduled reporting
	Specific Threshold 2 • Three consecutive exceedances of the 95th percentile value* from the baseline period (2009-2015) in routine monitoring results from a single monitoring port in MW09-03. *values provided in Table 2-9	**Notification** • Minto Management • Include in scheduled Water Use Licence monthly reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original results, or re-run samples if a laboratory error is indicated o Timing: initiate within 1 week of triggering SPT **Evaluation** • Review of groundwater monitoring data (including trend analysis) to be undertaken by a qualified professional, and appropriate recommendations to be developed o Review must consider the risk narrative (i.e. exceedance of surface water quality objectives as a result of groundwater flux) o Timing: initiate within 1 week of QA/QC review validating original results • Submit recommendations to regulator for review and approval o Timing: submit within 1 week of receipt of recommendations **Action** • Follow recommendations arising from review undertaken by qualified professional. o Timing: initiate implementation of recommendations within 1 month of receipt of approval from regulator • Increase monitoring frequency to monthly sampling of all monitored zones in the affected multi-level well for a period to be defined by qualified professional. • Trend analysis to be included in scheduled reporting

Table 2-10: Background concentrations and SPTs for the groundwater monitoring in the McGinty Creek watershed

Minto Creek		*Concentrations (mg/L)[1]*													
		Ag-D[3]	Al-D	As-D	Cd-D	Cr-D[3]	Cu-D	Pb-D[3]	Mo-D	Ni-D	Se-D	Zn-D	N-NO3	NH_4-N	Sulphate
MW09-03-01	95th Percentile	0.000016	0.0101	0.00081	0.000272	0.000575	0.00625	0.0003	0.026	0.0060	0.0015	0.022	0.28	0.133	38
	75th Percentile	0.000010	0.0065	0.00011	0.000075	0.000500	0.00155	0.0001	0.005	0.0021	0.0001	0.013	0.13	0.067	24
	Median[2]	0.000010	0.0045	0.00005	0.000022	0.000500	0.00031	0.0001	0.004	0.0015	0.0001	0.006	0.07	0.045	22
MW09-03-02	95th Percentile	0.000034	0.0095	0.00092	0.000272	0.000796	0.01080	0.0002	0.062	0.0026	0.0040	0.015	0.07	0.282	67
	75th Percentile	0.000018	0.0073	0.00074	0.000031	0.000500	0.00263	0.0001	0.018	0.0009	0.0005	0.010	0.03	0.230	7
	Median[2]	0.000010	0.0062	0.00067	0.000026	0.000500	0.00122	0.0001	0.017	0.0005	0.0002	0.008	0.01	0.210	1
MW09-03-03	95th Percentile	0.000010	0.0075	0.00014	0.000069	0.000500	0.00500	0.0003	0.018	0.0011	0.0004	0.011	0.54	0.058	13
	75th Percentile	0.000010	0.0047	0.00005	0.000023	0.000500	0.00247	0.0001	0.006	0.0005	0.0004	0.008	0.50	0.020	12
	Median[2]	0.000010	0.0025	0.00005	0.000015	0.000500	0.00174	0.0001	0.005	0.0005	0.0003	0.003	0.48	0.012	11

Notes:

1: For monitoring results where concentrations were below the analytical detection limits, a concentration of half the detection limit was adopted for calculation purposes.

2: For AMP purposes, the background concentration in groundwater at MW09-03 is defined as the median concentration observed in each port over the 2009-2015 baseline monitoring period.

3: For Ag-D, Cr-D and Pb-D, most 2009-2015 concentrations were at the limit of analytical detection, and as such the calculated 75th and 95th percentile values are skewed low.

2.5 Water Management

2.5.1 Description

The Minto Mine site has a positive water balance. Therefore, it is necessary to release water from site from time to time to prevent accumulation of excess water. The primary objective of Minto's water management strategy is to ensure that water can be released from site in a way that protects the water quality in Minto Creek. Details concerning water management for Phase V/VI are provided in the Minto Mine Phase V/VI Water Management Plan as amended from time to time (Minto Explorations Ltd., 2015).

The strategy can be summarized as follows:

- Runoff from developed mine areas (mine water) will be collected and stored in the Main Pit Tailings Management Facility (MPTMF) and the Area 2 Pit Tailings Management Facility (A2PTMF). Mine water will be used for ore processing.
- The site water balance will be used to define mine water inventory targets and targets for volumes to be released to Minto Creek. Inventory targets will be defined on an annual basis and reported in the annual water balance update.
- To the extent possible, water will be released from site by collecting and diverting discharge-compliant (clean) runoff to the water storage pond (WSP) and from there to Minto Creek.
- If collection, diversion and release of clean water does not move enough water off site then Minto has the option of treating and releasing mine water.

The water management strategy is able to deal with most foreseeable conditions that may be encountered though the mine development. However, certain unforeseen conditions may require an adaptive response as described below.

Based on observations and studies conducted for Phase V/VI licensing it is not expected that the Minto North Pit will fill to a surface spilling point during operations. It is a reasonable expectation given the location of the Minto North Pit within the watershed (located near the surface water divide between the McGinty Creek and Minto Creek catchments, with a small catchment reporting to the Minto North Pit).

2.5.2 Risk Narrative

The existing water treatment plant is not able to treat and discharge enough mine water, and as a result the mine water inventory exceeds the target.

2.5.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to water management are provided below in Table 2-11. The monitoring results that are evaluated and utilized for this component of the AMP are a requirement of the Water Inventory Tracking of the Water Management Plan (Minto Explorations Ltd., 2015). The monitoring data will be compared to the specific performance thresholds monthly (by the end of the month following the month in which the data were collected) - this corresponds with the existing monthly reporting schedule.

Table 2-11: Specific Indicators, Performance Thresholds and Responses for Water Management

Specific Indicators	Specific Performance Thresholds	Specific Responses
Water inventory in the Main Pit Tailings Management Facility or the Area 2 Pit Management Facility exceeds target inventory	**Specific Threshold 1** • Water inventory target is exceeded for a period of three months. The water storage capacity still exceeds 1,000,000 m^3.	**Notification** • Minto Management • Include in monthly report **Review** • Review site water balance • Review recent water management and water treatment practices **Evaluation** • Evaluate the water inventory targets. For example, how much water can be stored in the pits for how long? Can the inventory target safely be changed to accommodate the excess volume of water? **Action** • Develop plan to address the water excess inventory such that the target can be met within 6 months. The plan may include: o An adjustment of the target inventory, o Diverting more clean water to the WSP, o Modifying or expanding water treatment. o Trend analysis to determine when or if SPT2 will be triggered
	Specific Threshold 2 • The water storage capacity is less than 1,000,000 m^3.	**Notification** • Minto Management, SFN and YG Inspector. • Include in scheduled Water Use Licence reporting. **Review** • Review site water balance. • Review recent water management and water treatment practices. **Evaluation** • Evaluate the water inventory targets. For example, how much water can be stored in the pits for how long? Can the inventory target safely be changed to accommodate the excess volume of water? • Evaluate treatment requirements and determine if the exiting water treatment plant has sufficient capacity to meet the requirements. **Action** • Immediately develop and implement a plan to address the lack of storage capacity such that the capacity can be restored prior to subsequent freshet. o The plan may include, modifying or expanding the water treatment plant. o Plans to bring mobile treatment equipment to site may be considered. o Trend analysis to determine when or if SPT3 will be triggered

Specific Indicators	Specific Performance Thresholds	Specific Responses
	Specific Threshold 3 • The water storage capacity is less than 500,000 m^3.	**Notification** • Minto Management, SFN, YG Inspector and Regulators. • Include in Water Use Licence reporting. **Review** • Review site water balance. • Review recent water management and water treatment practices. **Evaluation** • Evaluate treatment requirements and determine if the existing water treatment plant has sufficient capacity to meet the requirements. **Action** • Immediately make plans to bring mobile treatment equipment to site, if existing plant does not have sufficient capacity. • Conduct trend analysis to determine duration until spilling

2.5.4 Freeboard Thresholds for Water Storage Facilities

The water storage facilities authorized under the WL include the Water Storage Pond, the Mill Water Pond, the Ridgetop North Pit, the Main Pit and the Area 2 Pit. With the exception of the Ridgetop North Pit, all of these facilities have documents associated with their design.

Table 2-12: Freeboard Limits for Storage Facilities

Storage Facility	Freeboard limit (m) from spill elevation
Water Storage Pond	1
Main Pit Tailings Management Facility	2
Area 2 Pit Tailings Management Facility	5

All of the water storage facilities water levels and volumes are managed through permanent pumping and piping systems.

The mill water pond was decommissioned in 2016 and is therefore not included in this AMP.

2.6 Physical Stability

2.6.1 Description

The physical stability of the waste rock, tailings and water storage facilities are monitored according to the Physical Monitoring Plan, which forms part of the Environmental Monitoring, Surveillance and Reporting Plan (Minto Explorations Ltd., 2016). The document describes the inspection and instrumentation data collection frequencies, instrument locations, installation details, as well as the data collection procedures.

The purpose of the monitoring program is to identify physical changes to the conditions of the facilities which may lead to future instability and to allow the mine to mitigate these conditions prior to any occurrence of instability. The facilities have been separated into two sets of geotechnical thresholds and response criteria (Table 2-13).

Table 2-13: Physical Stability Categories

Category	Facility
1	• Dry Stack Tailings Storage Facility and Mill Valley Fill Extension (Stage 1 and 2) • Southwest Waste Dump • South Wall Buttress / Main Pit Dump
2	• Main Waste Dump and Main Waste Dump Extension • Reclamation Overburden Dump • Ice-Rich Overburden Dump • Water Storage Pond Dam

Category 1 facilities are founded in areas of ice-rich periglacial foundations that have previously experienced deep seated foundation movement. The Mill Valley Fill Extension (MVFE) and South Wall Buttress (SWB) are designed to mitigate movements in the Dry Stack Tailings Storage Facility (DSTSF) and Main Pit South Wall areas,

respectively. Additional monitoring inspection and response requirements for the DSTSF are detailed in the Operations, Maintenance, and Surveillance Manual (OMS) for the facility (Minto Explorations Ltd., 2014).

Category 2 facilities consist of all the remaining waste rock dumps and the Water Storage Pond Dam. These waste dumps are located in areas with good foundation conditions that avoid areas underlain by ice-rich overburden. Additional monitoring inspection and response requirements for the Water Storage Pond Dam are detailed in the Operations, Maintenance, and Surveillance Manual (OMS) for the facility (Tetra Tech EBA, 2014).

2.6.2 Risk Narrative

A mass failure of one of the waste facilities has the potential to endanger the health and safety of site employees or visitors, or lead to an increase in contaminant loadings from the mine and subsequent adverse effects to aquatic resources in the receiving environment (lower Minto Creek).

2.6.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to Category 1 and Category 2 Facilities are provided in Table 2-14 and Table 2-15, respectively. The monitoring results that are evaluated and utilized for this component of the AMP are a requirement of the Physical Monitoring Program of the EMSRP (Minto Explorations Ltd., 2016). The monitoring data will be compared to the specific performance thresholds monthly (by the end of the month following the month in which the data were collected) - this corresponds with the existing monthly reporting schedule.

Table 2-14: Specific Indicators, Performance Thresholds and Responses for Category 1 Facilities

Specific Indicators	Specific Performance Thresholds	Specific Responses

• Mass movement indicated by monitoring of geotechnical instrumentation
• Visual observations of physical damage
• Visual observations of evidence that could suggest mass movement
• Occurrence of seismic events
	Specific Threshold 1
	
	• Observation of unusual occurrence including:
	  • tension cracks, settlement, or sloughing;
	  • a seismic event that exceeds the 1:475 return period event[1];
	  • abnormal seepage from any area of the slopes;
	  • increased turbidity from seepage; or,
	  • physical damage.
	
	This threshold applies to all Category 1 facilities.
		Notification
		• Mine Manager
		• Geotechnical Engineer/EIT
		• Chief Engineer
		• Include in annual report
		
		Review
		• Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data.
		
		Evaluation
		• Geotechnical Engineer/EIT to compare recent monitoring results against older results for additional evidence of instability.
		
		Action
		• Inspect the area for any other signs of instability.
		• Follow any recommendations of the Geotechnical Engineer/EIT. At a minimum, the Engineer/EIT will consider the need for:
		  o An increase in the frequency of routine inspections and monitoring.
		  o Additional inspection, instrumentation, monitoring, or analyses.
		
		• If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.

	Specific Threshold 2 • One survey hub or inclinometer reading indicating an increase in the movement rate greater than the long-term trend and outside the range of instrumentation error. This threshold applies to all Category 1 facilities.	**Notification** • Geotechnical Engineer/EIT **Review** • Review existing instrumentation data. **Evaluation** • Geotechnical Engineer/EIT to compare recent monitoring results against older results and complete a trend analysis. **Action** • Retake reading. • If the reading was accurate, increase the survey hub or inclinometer frequency. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.

[1] This size of a seismic event would be felt by most people on site. It would shake buildings, and rattle or break dishes, hanging objects, etc. Earthquake information may also be found online at: http://www.earthquakescanada.nrcan.gc.ca/index-eng.php

Specific Indicators	Specific Performance Thresholds	Specific Responses
	Specific Threshold 3 • For DSTSF and MVFE piezometers 13-DSP-05a, 13-DSP-06, 15-DSP-07, and 15-DSP-08, and all SWD piezometers, an increase in piezometric pressures under unfrozen or thawing conditions such that Ru[2] exceeds 0.43. Conversions of Ru to equivalent water elevations for each instrument are contained in Table 2-15. • For MVFE2 piezometer 15-DSP-10, an increase in piezometric pressure under unfrozen or thawing conditions such that the equivalent water elevation is 3 m above the original ground surface. Or • Temperature greater than zero at a depth of 2 m below original ground *(all SWD ground temperature cables, and DSTSF ground temperature cables DST-10, DST-11, and DST-14 only)*	**Notification** • Mine Manager • Geotechnical Engineer/EIT • Chief Engineer • Include in annual report **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer/EIT to compare recent monitoring results against older results and complete a trend analysis. **Action** • Inspect the area for any other signs of instability. • If the piezometric pressure threshold is exceeded: immediately increase frequency to **twice-weekly** or as directed by the Engineer/EIT until determined unnecessary. • Follow any additional recommendations of the Geotechnical Engineer/EIT. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.
	Specific Threshold 4 • For DSTSF and MVFE piezometers 13-DSP-05, 13-DSP-06, 15-DSP-07, and 15-DSP-08, and all SWD piezometers, an increase in piezometric pressures under unfrozen or thawing conditions such that Ru exceeds 0.6. Conversions of Ru to equivalent water elevations for each instrument are contained in Table 2-15. • For MVFE2 piezometer 15-DSP-10, an increase in piezometric pressure under unfrozen or thawing conditions such that the equivalent water elevation is 10 m above the original ground surface.	**Notification** • Mine Manager • Geotechnical Engineer/EIT • Chief Engineer • Include in annual report **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer/EIT to compare recent monitoring results against older results and complete a trend analysis. **Action** • Inspect the area for any other signs of instability. • Immediately increase piezometric pressure monitoring and data review frequency to **daily** or as directed by the Engineer/EIT until determined unnecessary. • Follow any additional recommendations of the Geotechnical Engineer/EIT. At a minimum, the Engineer/EIT will consider the need for: o An increase in the frequency of routine inspections and monitoring. o Additional inspection, instrumentation, monitoring, or analyses. o Modifications to the waste placement/construction practices. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately. • Trend analysis will be included in scheduled reporting

[2] Ru is the pore water pressure coefficient which is the ratio of piezometric pressure to the overburden pressure. A pore water pressure ratio of 0.5 would be similar to the effect of a groundwater table at surface. Conversions from Ru to equivalent water elevation are contained in Table 2-16.
[3] Piezometer 13-DSP-5b exceeded this trigger on April 2015. A review of the data and stability analysis was completed and documented in the 2015 annual geotechnical inspection report and found no stability issue and recommended continued monthly monitoring.

Specific Indicators	Specific Performance Thresholds	Specific Responses
	Specific Threshold 5 • Three consecutive survey hub or inclinometer readings indicating an increase in the movement rate movement greater than the long-term trend. Or • Three consecutive survey hub readings indicating a change in horizontal direction of movement greater than 15 degrees from the long term trend. This threshold applies to all Category 1 facilities.	**Notification** • Mine Manager • Geotechnical Engineer/EIT • Chief Engineer • Include in annual report **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer/EIT to compare recent monitoring results against older results and complete a trend analysis. **Action** • Inspect the area for any other signs of instability. • Complete a ground survey of the area of interest to monitor any future displacement. • Immediately increase survey hub monitoring and data review frequency to **twice-weekly** or as directed by the Engineer/EIT until determined unnecessary. • Follow any additional recommendations of the Geotechnical Engineer/EIT. At a minimum, the Engineer/EIT will consider the need for: o An increase in the frequency of routine inspections and monitoring. o Additional inspection, instrumentation, monitoring, or analyses. o Modifications to the waste placement/construction practices, including discontinuation of loading. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately. • Trend analysis will be included in scheduled reporting.

Table 2-15: Specific Indicators, Performance Thresholds and Responses for Category 2 Facilities

Specific Indicators	Specific Performance Thresholds	Specific Responses

• Mass movement indicated by monitoring of geotechnical instrumentation
• Visual observations of physical damage
• Visual observations of evidence that could suggest mass movement
• Occurrence of seismic events
	Specific Threshold 1
	
	• Observation of unusual occurrence including:
	  • tension cracks, settlement, or sloughing;
	  • a seismic event that exceeds the 1:475 return period event;
	  • abnormal seepage from any area of the slopes;
	  • increased turbidity from seepage;
	  • physical damage.
	
	This threshold applies to all Category 2 facilities.
		Notification
		• Mine Manager
		• Geotechnical Engineer/EIT
		• Chief Engineer
		• Include in annual report
		
		Review
		• Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data.
		
		Evaluation
		Geotechnical Engineer/EIT to compare recent monitoring results against older results for additional evidence of instability.
		
		Action
		• Inspect the area for any other signs of instability.
		• Follow any recommendations of the Geotechnical Engineer/EIT. At a minimum, the Engineer/EIT will consider the need for:
		  o An increase in the frequency of routine inspections and monitoring.
		  o Additional inspection, instrumentation, monitoring, or analyses.
		• If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.

	Specific Threshold 2 **WSP Dam:** • One survey hub reading indicating an increase of movement outside range of instrumentation error. **All other Category 2 Facilities:** • Survey hub cumulative displacements between 150 mm and 500 mm.	**Notification** • Mine Manager • Geotechnical Engineer/EIT • Chief Engineer • Include in annual report **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer/EIT to compare recent monitoring results against older results for additional evidence of instability and complete a trend analysis. **Action** • Retake reading • If the reading is accurate, inspect the area for any signs of instability. • Immediately increase survey hub reading frequency to **twice-monthly** or as directed by the Engineer/EIT until determined unnecessary. • Follow any additional recommendations of the Geotechnical Engineer/EIT. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.

Specific Indicators	Specific Performance Thresholds	Specific Responses
	Specific Threshold 3 **WSP Dam:** • One piezometer reading outside of its long-term trend (in comparison to the reservoir pond elevation). **All other Category 2 Facilities:** • Survey hub cumulative displacements greater than 500 mm.	**Notification** • Mine Manager • Geotechnical Engineer/EIT • Chief Engineer • Include in annual report **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer/EIT to compare recent monitoring results against older results for additional evidence of instability and complete a trend analysis. **Action** • Inspect the area for any other signs of instability. • Complete a ground survey of the area of interest to allow for a stability assessment to be completed (if required by the Engineer), and to monitor any future displacement. • Immediately increase monitoring and data review frequency until determined unnecessary: o If survey hub threshold exceeded: increase frequency to **twice-weekly** or as directed by the Engineer/EIT. o If piezometric pressure threshold exceeded: increase frequency to **daily** or as directed by the Engineer/EIT. • Follow any additional recommendations of the Geotechnical Engineer/EIT. At a minimum, the Engineer/EIT will consider the need for: o An increase in the frequency of routine inspections and monitoring. o Additional inspection, instrumentation, monitoring, or analyses o Modifications to the waste placement/construction practices, including discontinuation of loading. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.
	Specific Threshold 4 **WSP Dam:** • Three consecutive survey hub readings indicating increase in movement outside range of instrumentation error.	**Notification** • Mine Manager • Geotechnical Engineer/EIT • Chief Engineer • Include in annual report **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature and survey data. **Evaluation** • Geotechnical Engineer/EIT to compare recent monitoring results against older results for additional evidence of instability and complete a trend analysis.

Specific Indicators	Specific Performance Thresholds	Specific Responses
		Action • Inspect the area for any other signs of instability. • Complete a ground survey of the area of interest to allow for a stability assessment to be completed (if required by the Engineer/EIT), and to monitor any future displacement. • Immediately increase survey hub reading frequency to **twice-weekly** or as directed by the Engineer/EIT until determined unnecessary. • Follow any additional recommendations of the Geotechnical Engineer/EIT. At a minimum, the Engineer/EIT will consider the need for: o An increase in the frequency of routine inspections and monitoring. o Additional inspection, instrumentation, monitoring, or analyses. o Modifications to the waste placement/construction practices, including discontinuation of loading. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.

Table 2-16: Equivalent Water Elevations for Piezometric Pressure Thresholds

Instrument	Facility	Threshold 1	Threshold 2	Threshold 3	Threshold 4	Threshold 5
DSP-05a	DSTSF	n/a	n/a	785.03 m	794.82 m	n/a
DSP-06a	DSTSF	n/a	n/a	784.18 m	791.49 m	n/a
DSP-06b	DSTSF	n/a	n/a	783.06 m	791.81 m	n/a
DSP-07 #1	DSTSF	n/a	n/a	777.89 m	781.13 m	n/a
DSP-07 #2	DSTSF	n/a	n/a	775.65 m	781.77 m	n/a
DSP-07 #3	DSTSF	n/a	n/a	773.69 m	782.33 m	n/a
DSP-07 #4	DSTSF	n/a	n/a	772.85 m	782.57 m	n/a
DSP-07 #5	DSTSF	n/a	n/a	772.29 m	782.73 m	n/a
DSP-07 #6	DSTSF	n/a	n/a	771.17 m	783.06 m	n/a
DSP-08 #1	DSTSF	n/a	n/a	774.32 m	786.20 m	n/a
DSP-08 #2	DSTSF	n/a	n/a	752.75 m	756.35 m	n/a
DSP-08 #3	DSTSF	n/a	n/a	751.35 m	756.75 m	n/a
DSP-08 #4	DSTSF	n/a	n/a	749.86 m	757.07 m	n/a
DSP-08 #5	DSTSF	n/a	n/a	748.46 m	757.47 m	n/a
DSP-08 #6	DSTSF	n/a	n/a	745.67 m	758.27 m	n/a
DSP-10	DSTSF	n/a	n/a	722.47 m	725.10 m	n/a
SDP-2A	SWD	n/a	n/a	846.40 m	847.90 m	n/a
SDP-2B	SWD	n/a	n/a	846.21 m	847.96 m	n/a
SDP-3A	SWD	n/a	n/a	859.10 m	861.52 m	n/a
SDP-3B	SWD	n/a	n/a	858.90 m	861.57 m	n/a
SDP-4A	SWD	n/a	n/a	860.47 m	861.46 m	n/a
SDP-4B	SWD	n/a	n/a	860.28 m	861.52 m	n/a

3 Reporting and Review

Reporting and review represent an essential part of the Adaptive Management Framework, as described in section 1.4.2, and reporting on adaptive management is included in the notification component of all specific responses described in section 2.

3.1 Monthly and Annual Reporting

Monthly reports are required to be submitted to the Yukon Water Board under Water Licence QZ14-031, and annual reports are required for submission under both the Water Licence and the Quartz Mining License. Both licenses require reporting on adaptive management.

Monthly reporting includes all activities carried out under the Adaptive Management Plan. The monthly report will include a comparison of monitoring results to AMP thresholds as well as status of responses implemented due to thresholds triggered.

Annual reporting will include summaries of all activities carried out under the Adaptive Management Plan, including a summary of the comparisons conducted monthly and any actions taken. The annual report will also include follow: water levels, a trend analysis (where applicable) for all parameters that triggered responses and a yearly comparison summary of all physical stability monitoring results to AMP thresholds.

The Annual Water Licence Report will also include a comparison of the data collected under this monitoring with predicted water quality estimates. This will be undertaken as part of the annual review of the Water and Load Balance Model.

3.2 Annual Review

The AMP may be modified when unexpected circumstances are encountered and the protocol is implemented or when additional understanding becomes available. An annual review of the AMP will take place prior to annual reporting, and Annual Reports will include a summary of proposed updates and revisions to the Adaptive Management Plan and include a revised Adaptive Management Plan, if warranted.

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Minto Mine

Closure Adaptive Management Plan

2018-01

Prepared by:
Minto Explorations Ltd.
Minto Mine
February 2018

Minto Mine Closure Adaptive Management Plan

AMP Version History

AMP Version	Issue Date	Description and Revisions Made
2014-01	June 2014	First Issue
2016-01	August 2016	Revisions made as per requirements of WUL QZ14-031 and QML-0001
2017-01	February 2017	Revisions made to reflect finalization of closure water quality objectives with SFN, with associated updates and adjustments
2018-01	February 2018	Revisions made based on YWB review and subsequent information requests.

2017-01 Revisions

AMP Section	Description of Revision
1	Included a description of all three closure periods. Included a recommendation to revise the AMP prior to implementation.
2.1	Included an introduction paragraph which provides context by explaining how the surface water quality AMP components are integrated. Also, included Table 2-1, which clarifies what AMP component applies to a given closure period.
2.1.1.1	Included clarification regarding what closure period the Mine Site AMP Component applies.
2.1.1.3	Revised all of the values in Table 2-2 (formerly 2-1) so that they actually represent maximum predicted concertation for the Post Closure II period. Revised Table 2-3 (formerly 2-2) to include applicable closure period to each specific threshold and revised wording to improve clarity.
2.1.2.1	Included explanation for when the CWTS AMP component applies.
2.1.2.3	Revised Table 2-5 (formerly 2-4) to include applicable closure period to each specific threshold.
2.1.3.1	Significant revisions made to describe how the Minto Creek AMP component will apply to the PCI period.
2.1.3.2	Significant revisions made to separate out risk narrative for PCI and PCII periods.
2.1.3.3	Significant revisions made to incorporate a PCI specific AMP component for Minto Creek. Includes the addition of Table 2-6 for the PCI period. Significant revisions made to the PCII specific AMP component to incorporate the finalized WQOs. Table 2-7 (formerly Table 2-5) revised with WQOs. Revisions specific threshold 1 of Table 2-8 (formerly Table 2-6). Revisions to Table 2-9 (formerly Table 2-7) to incorporate central tendency WQO, revised resample frequency in specific threshold 1 response, included applicable closure period to each specific threshold and revised specific threshold 5.
2.1.4	Significant revisions to match what is in the updated operational AMP (2016-02).
2.2.1.3	Significant revisions to match what is in the updated operational AMP (2016-02).

2018-01 Revisions

AMP Section	Description of Revision
1	Corrected errors in Table 1-2
2.1.1	Added details on the investigation/evaluation in response to SPT-1 in Table 2-3.
2.1.2	Included specific definition in SPT-3 in Table 2-5.
2.1.3	Included details on the application of trend analysis (both annual and the proposed monthly methods) in Table 2-6. Corrected errors in Table 2-7. Included trend analysis and more details on investigation/evaluation elements in Table 2-8.
2.1.4	Specified a 12-month moving average for comparison with the CTE in Table 2-11.
2.3.1	Included a list of specific facilities to which the Water Management Infrastructure AMP applies.
2.5.2	New section, AMP for Minto North Pit Water Levels

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1 Introduction

Minto Explorations Ltd. (Minto), a wholly owned subsidiary of Capstone Mining Corp. (Capstone), owns and operates the Minto copper mine. Minto Mine is located within Selkirk First Nation (SFN) Category-A Settlement Land (Parcel R-6A), and is approximately 240 km northwest of Whitehorse, Yukon Territory. The Minto mine commenced commercial operations in October 2007.

As an outcome of the Phase V/VI mine licensing processes, Minto was required to update the mine Reclamation and Closure Plan (RCP) for the updated mine plan and with additional information outlined in its Quartz Mining License QML-0001 (QML) and Water Use Licence QZ14-031 (WUL) and approval letters. Both authorizations required the updating of the Adaptive Management Plan (AMP) for the closure and post-closure periods. AMPs are tools used to address uncertainty or conditions beyond those expected in reclamation and closure. AMPs outline a range of possible but unexpected outcomes and the responses that will be undertaken to curb possible negative impacts associated with these unexpected situations.

Closure failure modes and effects analyses (FMEA) have been conducted to determine both the probability and severity of postulated component failures at the Minto mine site and the associated infrastructure and environmental management systems. The FMEAs have been used to determine potential failure modes with the greatest risk rating, and identified components and scenarios where uncertainty and attendant risk could be reduced through adaptive management.

Mining activities are highly managed operations, with very prescriptive and detailed management plans required for both operational control and regulatory approval. More mature mines such as Minto have management plans which benefit from the operational experience at the site, and uncertainty in the range of conditions expected is reduced through this operational experience.

Minto has developed and implemented an operational Adaptive Management Plan to address uncertainty in site performance during the operational period (MEL, 2016). Uncertainty is reduced in the operational period through the development and implementation of operational management plans. Minto has developed a number of operational management plans which describe the management and response actions for expected conditions at the site during operations. These plans currently include:

- Solid Waste Management Plan;
- Environmental Monitoring, Surveillance, Reporting Plan;
- Wildlife Protection Plan;
- Spill Contingency Plan;
- Sediment and Erosion Control Plan;
- Mine Development and Operations Plan;
- Underground Mine Development and Operations Plan;
- Mill Operations Plan;
- Water Management Plan;

- Tailings Management Plan;
- Waste Rock and Overburden Management Plan;
- Emergency Response Plan; and
- Heritage Resources Protection Plan.

For the closure and post closure periods, planning and management of site activities are guided by the Minto Mine Reclamation and Closure Plan (RCP). This includes the identification of distinct closure periods with defined activities and performance expectations:

1. Active Closure (AC) - includes the implementation and construction of the large majority of the selected closure measures. Closure measures completed during this period include, but are not limited recontouring, soil placement, closure water conveyance construction, revegetation, construction of passive water treatment facilities and partial decommissioning of the Water Storage Dam, demobilization and demolition. Further details regarding the planned closure measures can be found in Section 7 of the RCP. During the AC period, operational water quality objectives/effluent standards and AMP will still apply and active treatment is expected to be the primary source of water treatment. The AC period is expected to span three years but may be completed sooner as closure measures are completed.
2. Post Closure I (PCI) - The PCI period is intended to provide time which allows the closure measures to establish and assessments to be completed on the performance of the chosen closure measure. During this period, maintenance on soil covers will be ongoing as vegetation begins to establish. Water conveyance features will be commissioned along with the passive treatment system. The passive treatment system is expected to take two years to reach design performance expectations. A pump back system will be established from the passive treatment system to the water treatment plant as there is potential that water quality will not meet effluent standards without active treatment. During PCI, operational water quality objectives/effluent standards will still apply. Passive treatment will be commissioned over the first two years of PCI and will become the primary water treatment system utilizing active treatment as required to meet operational water quality objectives and manage site water inventory. During the PCI period the Operational AMP and Closure AMP will both apply to specific components and are identified in this AMP.
3. Post Closure II (PCII) - is intended as a confirmatory period which will monitor that the closure measures continue to perform as expected. PCII is primarily a monitoring phase with maintenance activities as required. During PCII the closure water quality objectives will apply and passive treatment will be the primary water treatment onsite. The Closure AMP will be the sole AMP in PCII and will cover all of the components detailed in this plan.

This AMP is intended to provide a framework for responses to conditions beyond those expected and identified in the RCP for primarily the Post Closure I and II (PCI and PCII) periods. It has been constructed to meet the licensing requirements under both main authorizations, and has benefited from an initial adequacy review by the Yukon Water Board (version 2017-01), with updates included based on that review in this version 2018-01.

Given the complexities of this AMP and the duration until implementation, it is recommended that this plan be reviewed and revised as required prior to implementation.

1.1 Adaptive Management Overview

Adaptive management is an approach to environmental management that is appropriate when mitigation measures may not function as intended or when broad-scale environmental change is possible. Adaptive management plans are precautionary in nature, and provide a level of security in long term environmental planning. Adaptive management plans also allow for the inclusion of improved science into mitigation measures as they are continually revised.

Adaptive management has been evolving since its emergence in the 1970s. Adaptive approaches include an ability to incorporate knowledge into the management plan as the knowledge is gleaned and circumstances change. Embedding adaptation into environmental plans involves thinking about how the results of monitoring will change management actions. Eberhard et al. (Eberhard, et al., 2009) described the categories of knowledge that may trigger changes to water quality management plans; system understanding, measuring progress and anticipating changes. These categories allow for the inclusion of knowledge and adaptation of management to changed conditions. Adaptive management plans are a way to accept uncertainties and build a structured framework to respond to changing conditions.

Adaptive management conducts a flexible path with actions to take when specific triggers occur. AMPs are a formalization of a plan for performance monitoring and project re-evaluation in the future. The general structure of adaptive management can be described by the following steps:

1. Identify risk triggers associated with vulnerabilities or uncertainties;
2. Quantify impacts and uncertainties;
3. Evaluate strategies and define implementation path that allows for multiple options at specific triggers;
4. Monitor the performance and critical variables in the system; and
5. Implement or re-evaluate strategies when triggers are reached.

Although there are no widely used AMP terms, the steps listed above are representative of typical AMP processes. Within AMPs, triggers provide decision points in a stepwise decision-making framework that identifies how and when management action should be taken. A key characteristic of adaptive management is monitoring, which is used to advance scientific understanding and to adjust management policies in an iterative process. Adaptive management is a rigorous method for addressing uncertainties in ecosystem management.

1.2 Adaptive Management Plan Objectives

An AMP is a management tool wherein a framework is provided to make quick and effective decisions to guide responses to unforeseen events. This document identifies areas of uncertainty within the post closure phases of the Minto Mine life and provides an AMP framework for each. For each component the AMP describes thresholds, triggers and responses to underperforming elements or emerging risks within the component. The steps laid out in the AMP framework are precautionary, and therefore they provide the confidence that action will be taken before adverse environmental impacts are observed.

Response planning, and results for anticipated events are contained within site management plans while AMPs guide responses to unforeseen or contingency events. This AMP provides a framework to guide responses to unanticipated monitoring results and to potential but low probability events where uncertainty exists.

It is difficult to predict the specific environmental condition that may arise which requires a response from management and, therefore, the AMP does not provide specific detailed descriptions of responses to a situation. The AMP provides a range of possible responses to use as a guide to respond to specific environmental conditions encountered. Management should use the information provided in the AMP and undertake the appropriate response.

1.2.1 Updated AMP Objectives

The 2016-01 version of the AMP was updated to include the requirements of the QML and WUL.

The QML was issued December 18, 2014 and was accompanied by a letter entitled outlining the components required in an Adaptive Management Plan (Yukon Government - Energy, Mines and Resources, 2014). The AMP includes the requirements outlined in the letter, which have also been addressed in the Operational AMP (the QML approval letter is non-specific as to the mine phase, so the requirements have been included in both AMPs). The QML approval letter requirements and where the items are addressed in the AMP are presented in the following concordance Table 1-1.

Table 1-1: QML Approval Letter Concordance Table

1. Adaptive Management Plan An adaptive management plan must include:	Section addressed
(a) key pathways and events that lead to uncertainty about site performance, including but not limited to:	1.3 2.1 – 2.5
o water quality	2.1
o physical stability	2.2
o cover systems	2.3.1
o water treatment, and	2.1, 2.5
o water management	2.5
(b) detailed descriptions of effective indicators of performance, monitoring programs, triggers for adaptive management responses, and actions that will be taken for each event or pathway. Triggers and actions should be defined and described for various levels of response, initially requiring investigation of causes and proceeding to implementation of appropriate contingency measures; and	2.1 – 2.5
(c) detailed descriptions of procedures for investigating causes of adaptive management trigger exceedances, and corrective actions should be provided. These should demonstrate that actions can be taken prior to causing unacceptable effects	2.1 – 2.5

The WUL was issued August 5, 2015 and the requirements for the Closure AMP are outlined in clause 110 e) (Yukon Water Board, 2015).

Clause 110 e) details and the sections in the AMP where these are addressed are summarized in Table 1-2.

Table 1-2: QZ14-031 Concordance Table

110) The updated RCP shall include, but not necessarily be limited to, the following: e) an updated Closure AMP including, but not necessarily be limited to, the following:	Section addressed
i) numerical values of thresholds for each parameter for each specific threshold for each AMP component	all
ii) consideration of statistical trend analysis to provide future forecasting of potential exceedances;	2.1
iii) thresholds for action and responses related to potential poor performance of the Main Pit Dam; and	n/a[1]
iv) management measures to responds to changing conditions in:	2.1
• McGinty Creek	2.1.4
• water quality at internal monitoring locations	2.1.1
• pit water quality	2.1.1
• passive treatment system performance; and	2.1.2
• water levels in open pits	2.5.2

[1] Main Dam is no longer part of the proposed Mine Plan

1.3 Adaptive Management Plan Approach

In addition to the conclusions drawn from research, the approach presented in this AMP follows the Environmental Code of Practice for Metal Mines, Section 4.1.17 on Adaptive Management:

"Mine owners/operators should use adaptive management methods to revise and refine the environmental management strategy. Adaptive management should consider a wide range of factors, including:

- The results of environmental audits or other evaluation activities;
- The results of environmental monitoring;
- The results of monitoring of the performance or condition of environmental infrastructure, such as containment structures, water management systems or treatment facilities;
- Technological developments; and
- Changing environmental conditions." (Environment Canada, 2009)

1.3.1 AMP Components

The following AMP components have been identified as having the potential for unexpected conditions during the operational period for which the Operational Management Plans may not provide adequate mitigation against potential effects to the environment or human health and safety, therefore, these components have been carried over into the AMP for the closure periods:

- Surface water quality;
- Groundwater quality;
- Water Management Infrastructure;
- Physical Stability; and
- General Reclamation Measures

The specific AMP framework for these components is described in Sections 2.1 through 2.5.

1.3.2 AMP Framework

The AMPs for each component are laid out using a common element approach to create consistency in implementation of the AMP protocol for all components as illustrated in Figure 1-1. The common elements are:

1. Description of the component
   - *Description* - description and understanding of the component leads to risk narrative and specific performance thresholds.
   - *Risk Narrative* describe the possible environmental impacts and environmental conditions that implementation of the AMP will prevent.
2. Monitoring the component

- *Specific Indicators* are the environmental or physical parameters to be monitored and assessed. Specific indicators are measurable or observable, and are indicative of changes from the designed or expected condition.
- *Monitoring Requirements* describes the monitoring regime for the component including frequency, type of data required and interpretation of results – outlined in the Minto RCP document.
- *Specific Performance Thresholds* define the conditions, in terms of specific indicators, when action is triggered. Performance thresholds are staged to accommodate levels of concern and a diversity of actions. To the extent possible, specific performance thresholds will include early warning thresholds.

3. Responding to unexpected conditions of the component

   Specific Responses are staged according to specific performance thresholds describes the actions to be implemented if specific performance thresholds are crossed. They are provided in the following categories:

   a) Notification

   b) Review

   c) Evaluation

   d) Action

4. Annual Reporting and Review

   Annual Reporting reflects annual changes made to the AMP as the site conditions change. The AMP should be modified whenever unexpected circumstances are encountered and the protocol is implemented or when additional proven science or technology becomes available. The annual review will include a review of the relevant monitored data and AMP elements. Updates, amendments, performance thresholds crossed, and trigger(s) activated will be provided to the appropriate governmental (including SFN) organizations as required and will be part of the annual report.

   Additional reporting is described further in section 3.

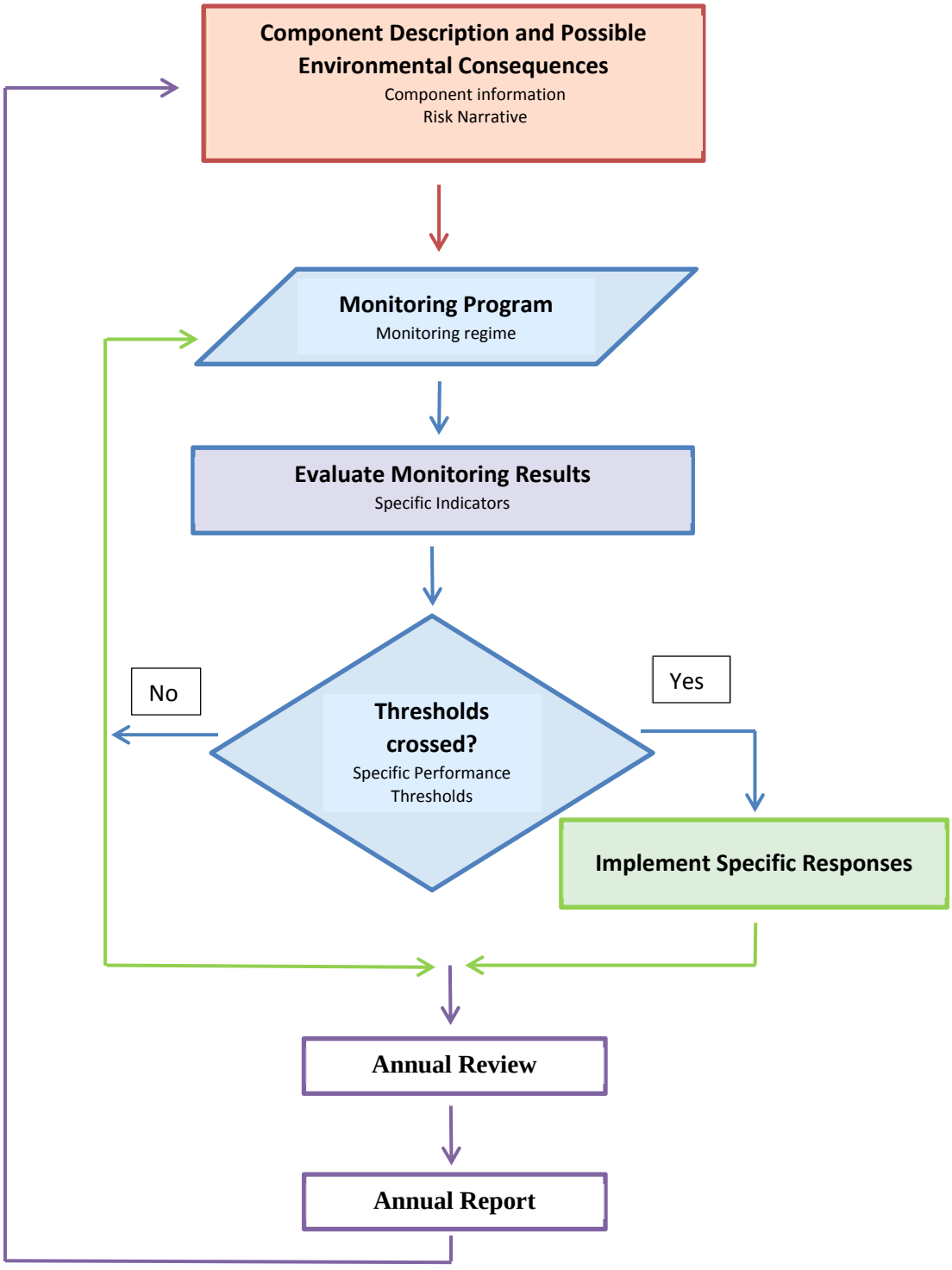


Figure 1-1: Sequential Components of the AMP (Adapted from AECOM 2010)

2 Adaptive Management Plans for Mine Components

2.1 Surface Water Quality

The Surface Water Quality section of this AMP aims to integrate the mine site (Area 2 and Main Pits) water quality with the performance of the CWTS to reach the overall goal of meeting water quality objectives (WQO) in Minto Creek's receiving environment (W2). The McGinty Creek AMP component is also included but does not have the same source control uncertainty as Minto Creek and is therefore not integrated with other AMP components. The Surface Water Quality components are intended to address the uncertainty that arise in specific closure periods and are therefore not completely inclusive of PCI and PCII closure periods. The applicable closure periods will be identified to the corresponding components in the sections below. A summary of the closure periods and corresponding AMP components is broken down in Table 2-1 below.

Table 2-1: Surface water AMP components and corresponding closure period

AMP Component	Applicable Closure Period	Exceptions
Mine Site	PCII	No exceptions
Constructed Wetland Treatment Systems (CWTS)	PCI and PCII	Applies after the CWTS commissioning period only
Minto Creek	PCI and PCII	Only applies in PCI after the CWTS is preforming as expected
McGinty Creek	PCI and PCII	No exceptions

2.1.1 Mine Site

2.1.1.1 Description

During the closure and post-closure period, the management of runoff and site water will become increasingly more passive, as described in the RCP. Attention to the quality of site runoff and surface water at some key locations will be important indicators of reclamation success of site facilities, and will provide an indication of potential water quality issues upstream of the passive treatment installation at the former water storage pond footprint. In particular, attention to the water quality in the two upstream pit outflows (Main Pit – W12, Area 2 Pit – W45) will provide site managers with indications of water quality that may lead to passive treatment performance problems, and possibly to water quality objectives not being attained further downstream in the receiving environment of Minto Creek (W2).

As active water treatment will be employed and utilized in Active Closure as well as available during the PCI period, with a high degree of certainty around effluent quality, the AMP framework for the mine site facilities below are relevant to the post closure PCII period only.

2.1.1.2 Risk Narrative

Increasing concentrations of contaminants from closed site facilities and flooded open pits lead to passive treatment challenges, exceedance of downstream water quality objectives, and/or adverse effects to aquatic resources in the receiving environment (lower Minto Creek).

2.1.1.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to water quality and the monitoring program are provided below in Table 2-3, and the specific thresholds are identified in Table 2-2. The monitoring results that will be evaluated and utilized in this component of the AMP are outlined in Section 7.11 of the Minto Mine RCP (2018-01).

Table 2-2: Specific Performance Thresholds for Mine Site Stations

Parameter	Worst Case Maximum Predicted Concentration for Post-Closure II Period					
	W45 (Area 2 Pit)			W12 (Main Pit)		
	May	July	Sept	May	July	Sept
Ammonia - N, mg/L	0.79	0.77	0.77	0.13	0.12	0.12
Nitrite - N, mg/L	0.55	0.54	0.54	0.12	011	0.10
Nitrate - N, mg/L	30	29	29	12	11	10
Aluminum (dissolved), mg/L	0.42	0.42	0.42	0.37	0.36	0.35
Arsenic (dissolved), mg/L	0.0029	0.0029	0.0029	0.0027	0.0026	0.0026
Cadmium (dissolved), µg/L	0.00018	0.00018	0.00018	0.00018	0.00017	0.00017
Chromium (dissolved), mg/L	0.0033	0.0032	0.0032	0.0033	0.0031	0.0031
Copper (dissolved), mg/L	0.045	0.045	0.045	0.061	0.056	0.056
Iron (dissolved), mg/L	0.85	0.85	0.84	1.3	1.2	1.2
Lead (dissolved), mg/L	0.0013	0.0013	0.0013	0.0010	0.00095	0.00094
Molybdenum (dissolved), mg/L	0.049	0.048	0.048	0.016	0.015	0.015
Nickel (dissolved), mg/L	0.0046	0.0045	0.0045	0.0045	0.0042	0.0042
Silver (dissolved), mg/L	0.00011	0.00010	0.00010	0.00013	0.00012	0.00012
Selenium (dissolved), mg/L	0.023	0.022	0.022	0.0083	0.0076	0.0075
Zinc (dissolved), mg/L	0.017	0.017	0.017	0.017	0.016	0.016

Table 2-3: Specific Indicators, Performance Thresholds and Responses for Mine Site Monitoring Locations

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
Dissolved metal parameters in Table 2-2.	**Specific Threshold 1** Worst Case Maximum water quality prediction is exceeded **AND** Potential or existing challenges to passive treatment are identified as a result of pit water quality	**PCII**	**Notification** • Minto Management **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • Compare with previous results and evaluate if concentrations are trending upwards • review all other available monitoring results (e.g. W15) and assess whether results show evidence of related conditions. • test appropriate other pit inflow locations to help isolate the source • Depending on which pit and parameters show elevated concentrations, evaluate the scale (single station, site- wide, specific facility, etc.) **Action** • If result is validated, **AND** • concentrations are trending upwards significantly, then: o Re-sample within one month of triggering ST1
Dissolved metal parameters in Table 2-2.	**Specific Threshold 2** Subsequent exceedance of ST1 in subsequent sampling (resampled or routine) and evaluation confirms upward trend in concentrations **AND** Potential or existing challenges to passive treatment are identified as a result of pit water quality	**PCII**	**Notification** • Minto Management • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • All evaluation measures identified in ST1 **Action** • Increase sampling frequency to monthly • Engage qualified professional to evaluate potential for passive treatment challenges from pit outflow concentrations • Retain qualified expert to develop mitigation plan, including recommendations. This could include, for example, consideration of the use of batch treatment to mitigate water quality in pits.
Dissolved metal parameters in Table 2-2.	**Specific Threshold 3** Continued exceedance of ST2 in subsequent sampling **AND** Potential or existing challenges to passive treatment are identified as a result of pit water quality	**PCII**	**Notification** • Minto Management, SFN, YG Inspector • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol o Review laboratory QA/QC report o Verify original result, or re-run sample if laboratory error indicated **Evaluation** • All evaluation measures for ST2 **Action** • Implement recommendations from Mitigation Plan • Maintain monthly sampling frequency

2.1.2 Constructed Wetland Treatment Systems

2.1.2.1 Description

The RCP describes how surface water quality will be managed during the post closure period. This includes the construction and operation of a constructed wetland treatment system (CWTS) in the footprint of the existing water storage pond. Based on site specific research, it is expected that the CWTS will take at least 2 full years before the wetland will be performing as designed. Research to date, also outlined in the RCP, on wetland performance at the Minto site suggests that a constructed wetland treatment system can be expected to mitigate certain water quality constituent loading during some months of the year, particularly during base flow summer months. Although some routine maintenance is expected for this facility, there also exists some remaining uncertainty regarding how the wetland will perform in decreasing aqueous constituent loads.

Section 2.1.1 above outlines how contributing loads from upstream sources will be managed adaptively, and section 2.3 outlines how an adaptive management approach will be used to respond to changing conditions related to the physical integrity of the CWTS as a water management infrastructure component. This section outlines how unexpected changes in the performance of the CWTS related to aqueous constituent load reduction will be identified and responded to after the expected commissioning period (two years after construction).

The CWTS AMP framework will provide a response towards continuous improvement of the system during PCI period and will address unexpected performance in PCII. The CWTS will mature as a treatment installation during the first two years of PCI – the Commissioning Period – and after it will be transitioned to the primary water treatment component for the duration of PCI (as long as Operational WQO can be met) and for the entirety of PCII. The CWTS AMP framework below applies to the third year of PCI and through the entire duration of PCII.

2.1.2.2 Risk Narrative

Aqueous constituent load reduction does not meet expectations or design criteria, resulting in downstream water quality objectives not being met, and/or adverse effects to aquatic resources in the receiving environment (lower Minto Creek).

2.1.2.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to performance of the CWTS are provided below in Table 2-5. Specific thresholds are identified in Table 2-4. The monitoring results that will be evaluated and utilized in this component of the AMP are outlined in the Minto Mine RCP (2018-01) (Section 7.11).

Table 2-4: Specific Performance Thresholds for Mine Site stations

Specific Indicator	Specific Threshold Value	Basis/rationale for threshold value
CWTS Soil Redox Probe reading	-100 mV	Desired range for optimal performance is -250 to -100 mV. Performance can still be expected as high as -50, so -100 as threshold is conservative
Treatment rate coefficient		
Cadmium (dissolved)	0.078 (first order)	Treatment rate coefficients should be evaluated as an annual average, based on inflow and outflow concentrations, and retention time within the CWTS (formulae are provided in CTWS Design Document, appended to Minto RCP). The threshold values here are from the design basis, based on expected performance from reclamation research to date, with an added range of 20% to allow for variation.
Copper (dissolved)	0.060 (first order)	
Selenium (dissolved)	0.00012 (zero order)	

Table 2-5: Specific Indicators, Performance Thresholds and Responses for CWTS performance.

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
Soil Redox	**Specific Threshold 1** average of all probes trending upwards and out of ideal range (-250 to -100) over 3 consecutive sampling events **OR** 25% of all probes reading > -100 in any sampling event	**Year 3 of PCI and all of PCII**	**Notification** • Minto Management • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Validate redox results o Confirm water depth around probes in question **Evaluation** • Consider installation of duplicate probes to confirm readings • Consider sampling of soil in vicinity of probes in question for total organic carbon (TOC) and nutrients (NPK). **Action** • If probe readings confirmed and TOC is low, add organic carbon as directed by qualified expert • If probe readings confirmed and soil nutrients are low, add fertilizer as directed by qualified expert • If water depths less that design amount, mitigate as recommended by qualified expert
Treatment rate coefficient	**Specific Threshold 2** Target treatment rate coefficient for specific aqueous constituent is not achieved in any year of monitoring	**Year 3 of PCI and all of PCII**	**Notification** • Minto Management • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • Increase sampling frequency of influent/effluent to monthly (if not already implemented) • Sample between cells to isolate problem cell(s) • Consider analysis for interfering/facilitating issues (e.g. iron/Sulphur) **Action** • Retain qualified expert to develop mitigation plan, including recommendations
Treatment rate coefficient	**Specific Threshold 3** Target treatment rate coefficient for specific aqueous constituent is not achieved for **3 consecutive years** **AND** Applicable downstream water quality objectives have been exceeded as defined by: SPT-3 (or higher) is triggered from Minto Creek AMP in Table 2-9.	**Year 3 of PCI and all of PCII**	**Notification** • Minto Management, SFN, YG Inspector • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol o Review laboratory QA/QC report o Verify original result, or re-run sample if laboratory error indicated **Evaluation** • All evaluation measures for ST2 **Action** • Implement recommendations from Mitigation Plan

2.1.3 Minto Creek

2.1.3.1 Description

The majority of the Minto Mine footprint (with the exception of the Minto North pit) is in the headwaters of the catchment area for Minto Creek. The W2 monitoring point on lower Minto Creek near the Yukon River is beyond the final mine water discharge point (W50) controlled by Minto, and is subject to a broad range of influences from tributaries and catchment areas that are beyond Minto's control. However, Minto is committed to monitoring the water quality at W2 and responding to changing water quality in lower Minto Creek as appropriate. The AMP frameworks below compliment the closure water management plan at the mine site with a decision-based structure for ensuring that negative impacts to lower Minto Creek from mining activities are avoided.

The monitoring results that will be evaluated and utilized in this component of the AMP are outlined in section 7.11 of the Minto Mine RCP (2018-01). The Operational Water Quality Objectives will continue to be employed in the Active Closure and PCI periods. As such, the Minto Mine Operational AMP is relevant and will be utilized during the Active Closure and PCI periods. To address the uncertainty of meeting the post closure water quality objectives (PC-WQO) within the expected 5-year duration of the PCI period, a PCI AMP framework has been developed for Minto Creek and is intended to be used as a supplement to the Operational AMP.

The PCI period AMP framework was developed to guide continuous improvement towards meeting PC-WQO after the CTWS has reached the expected performance and will only apply during the PCI period. The portion of the Minto Creek AMP that applies to the PCII period is intended to respond to uncertainty regarding meeting both non-degradation and 50% of assimilative capacity WQOs.

In the sections below, the PCI period AMP framework and the PCII period AMP framework for Minto Creek have been separated to simplify what AMP frameworks apply to what closure period.

2.1.3.2 Risk Narrative

Post Closure I Period

The CWTS performs as designed, however, the water quality in the receiving environment is not trending towards meeting PC-WQO by the end of the expected PCI period.

Post Closure II Period

Increase in contaminant concentrations from the mine causes exceedance of the PC-WQO in the receiving environment (lower Minto Creek).

2.1.3.3 Indicator, Specific Performance Thresholds and Responses

Post Closure I Period

The indicator, performance thresholds and responses specific to continuous improvement of water quality during the PCI period are provided in Table 2-6. The AMP framework for the PCI period applies to the receiving environment of Minto Creek (W2) and utilizes the results of trend analysis of routine monitoring to guide responses to improving water quality towards meeting the post closure water quality objectives prior to the expected completion of the PCI period (5 years total). Specifically this AMP only applies after the CWTS has met the designed performance for a period of one year and water quality at W2 is not meeting post closure water quality objectives.

Table 2-6: Continuous improvement for water quality at W2 during the PCI period.

Indicator	Specific Performance Threshold	Stage	Specific Responses
Complete an annual Trend Analysis of receiving environment water quality data (Minto Creek Station W2) to determine if water quality meets or is trending towards meeting PC-WQOs. To be initiated after successful establishment of the CWTS (PCI Years 3,4,5 at minimum). And Complete an ongoing analysis with monthly W2 data to detect deviation from prediction limits.	Annual analysis shows trending towards meeting PC-WQO and will likely meet them before end of PCI.	**PCI**	**Action:** • Re-evaluate in subsequent years and confirm or repudiate previous year's results
	Annual analysis shows trending towards meeting PC-WQO but will likely not meet before end of PCI.	**PCI**	**Notification:** • Minto Management **Review:** • Follow QA/QC investigative protocol: - Review laboratory QA/QC report - Validate original result, or re-run sample if a laboratory error is indicated **Evaluation:** • Review site water quality results - e.g. does pit water quality exceed expected concentrations? • Compare with W50 and/or CWTS results. If comparison indicates elevated site concentrations are responsible for exceedance; and validation confirms original result, then: - Engage qualified expert(s) to develop mitigation plan **Action:** • Review results of investigations/evaluations and implement recommendations if appropriate. • Re-evaluate in subsequent years and confirm or repudiate previous year's results
	Annual analysis shows water quality stable, neither trending towards nor away from PC-WQO OR Monthly Analysis shows 2 or more consecutive results above the upper prediction limit.	**PCI**	**Notification**: • Minto Management, SFN, YG Inspector • Include in scheduled Water Use Licence reporting **Review:** • Follow QA/QC investigative protocol: - Review laboratory QA/QC report - Validate original result, or re-run sample if a laboratory error is indicated **Evaluation:** • Review site water quality results - e.g. does pit water quality exceed expected concentrations? • Compare with W50 and/or CWTS results. If comparison indicates elevated site concentrations are responsible for exceedance; and validation confirms original result, then: - Engage qualified expert(s) to develop mitigation plan **Action:** • Review results of investigations/evaluations and prepare recommendations if appropriate. • Engage SFN and Regulators • Re-evaluate in subsequent years and confirm or repudiate previous year's results
	Annual analysis shows trending away from meeting PC-WQO OR Monthly Analysis shows 3 or more consecutive results above the upper prediction limit.	**PCI**	**Notification:** • Minto Management, SFN, YG Inspector • Include in scheduled Water Use Licence reporting **Review:** • Follow QA/QC investigative protocol: - Review laboratory QA/QC report - Validate original result, or re-run sample if a laboratory error is indicated **Evaluation:** • Review site water quality results - e.g. does pit water quality exceed expected concentrations? • Compare with W50 and/or CWTS results. If comparison indicates elevated site concentrations are responsible for exceedance; and validation confirms original result, then: - Engage qualified expert(s) to develop mitigation plan **Action:** • Review results of investigations/evaluations and prepare recommendations if appropriate. • Engage SFN and Regulators • Re-assess Passive treatment option • Re-evaluate in subsequent years and confirm or repudiate previous year's results

Post Closure II Period

Indicators, performance thresholds and responses specific to water quality and the monitoring program are provided below in Table 2-8. Specific thresholds for the PCII period AMP framework are identified in Table 2-7. This portion of closure AMP for Minto Creek water quality utilizes the post closure water quality objectives (WQOs) at station W2 as the basis for specific thresholds.

To apply these WQOs as the basis of specific performance thresholds in this plan, the following abbreviations, definitions and assumptions are being used:

BGWQ: Background Water Quality, as compiled and outlined in Appendix A (Minnow, 2016) of Appendix 1-3 (Surface Water Quality) of Appendix H (Site Characterization Plan) of the RCP 2018-01.

ND: Non-degradation of water quality, statistically unchanged from BGWQ

NDIDPE-WQO: Non-degradation Maximum water quality objective, as defined by the 95th percentile of the BGWQ data set

NDCT-WQO: Non-degradation Central Tendency water quality objective, as defined as the 95th percentile of the annual medians of the BGWQ data set.

50IDPE-WQO: 50% of assimilative capacity Individual Data Point Evaluator water quality objective, as calculated using the 95th percentile of background concentrations and the Post Closure WQO formula outlined in Section 3.1.3.1 of RCP 2018-01.

50CT-WQO: 50% of assimilative capacity Central Tendency water quality objective, as calculated using the NDCT-WQO as background and the Post Closure WQO formula outlined in Section 3.1.3.1 of RCP 2018-01.

75%-STI: Specific Threshold One value, as calculated using 75 percent of the 50IDPE-WQO and is intended to be an early warning threshold for the 50% of assimilative capacity WQO.

Table 2-7: Specific Performance Thresholds for Surface Water Quality in Lower Minto Creek (W2)

Parameter	Non-Degradation - Individual Data Point Evaluator	Non-Degradation - Central Tendency	75% of 50% Assimilative Capacity Individual Data Point Evaluator	50% Assimilative Capacity - Individual Data Point Evaluator	50% Assimilative Capacity - Central Tendency
	Water Quality Objective	Water Quality Objective	STP-1	Water Quality Objective	Water Quality Objective
	(NDIDPE-WQO)	(NDCT-WQO)	Table 2-8	(50IDPE-WQO)	(50CT-WQO)
Ammonia - N, mg/L	0.12	0.06	0.14	0.18	0.13
Nitrite - N, mg/L	0.04	0.03	0.04	0.05	0.04
Nitrate - N, mg/L	0.22	0.15	3.5	4.7	4.6
Aluminum (dissolved), mg/L	0.053	0.032	0.058	0.077	0.066
Arsenic (dissolved), mg/L	0.0011	0.0009	0.0023	0.0031	0.0029
Cadmium[1] (dissolved), µg/L	0.00005	0.000044	0.00017	0.00022	0.00019
Chromium (dissolved), mg/L	0.0011	0.0010	0.0008	0.0011	0.0010
Copper (dissolved), mg/L (when [DOC] @ W2 >10 mg/L)	0.0054	0.0029	0.010	0.013	0.011
Copper (dissolved), mg/L (when [DOC] @ W2 ≤10 mg/L)	0.0054	0.0029	0.0069	0.0092	0.0079
Iron (dissolved), mg/L	0.91	0.57	0.75	1.0	0.84
Lead (dissolved), mg/L	0.00025	0.00014	0.0016	0.0021	0.0021
Molybdenum (dissolved), mg/L	0.0015	0.0011	0.028	0.037	0.037
Nickel (dissolved), mg/L	0.0020	0.0016	0.042	0.056	0.056
Selenium (dissolved), mg/L	0.00050	0.00046	0.0010	0.0013	0.0012
Silver (dissolved), mg/L	0.00001	0.00001	0.000041	0.000055	0.000055
Zinc (dissolved), mg/L	0.008	0.005	0.014	0.019	0.017

[1]Cadmium value calculated using 95th percentile hardness (225.2 mg/L) in Operational Stage WQO calculation. Actual objective for comparison with monitoring results will be calculated with observed hardness.

Table 2-8 below presents the specific thresholds and actions proposed to respond to changing water quality, i.e. deviation from a non-degradation condition, in lower Minto Creek, as measured at water quality station W2.

Table 2-8: Specific Indicators, Performance Thresholds and Responses for Surface Water Quality at Station W2 in lower Minto Creek during Post Closure II, using Non-Degradation Water Quality Objectives.

Specific Indicators	Specific Performance Thresholds	Specific Responses
Aqueous Concentrations at Station W2 for parameters with water quality objectives (See Table **2-7** above for specific threshold values)	**Specific Threshold 1** • Exceedance of 2 consecutive NDIDPE-WQOs **OR** • Exceedance of NDCT-WQO once	**Notification** • Minto Management • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • Review site water quality results – e.g. does pit water quality exceed expected concentrations? • Conduct a trend analysis • Compare with site water quality results, including W3, CWTS results, potential groundwater inflows, and contributions from the bypass channel if applicable. If comparison indicates either elevated site concentrations or poor CWTS performance are responsible for exceedance; and validation confirms original result, then: o Engage qualified expert(s) to develop mitigation plan **Action** • Review results of investigations/evaluations and prepare recommendations if appropriate. This could include a recommendation that certain parameters be immediately subject to the AMP framework in Table 2-9 – 50%AC-WQOs if concentrations are determined by a qualified expert are unlikely to achieve ND-WQOs. • if trend analysis suggests a continued (sustained) exceedance of Threshold 1 within one year, then initiate actions from SPT-2 • Actions will continue until performance thresholds are no longer exceeded.
	Specific Threshold 2 • Continued Exceedance of Threshold 1 in subsequent sampling	**Notification** • Minto Management, SFN • Include in scheduled Water Use Licence reporting **Review** • All review measures for ST1 **Evaluation** All evaluation measures from ST1 **Action** • Implement recommendations from mitigation plan • Actions will continue until performance thresholds are no longer exceeded.
	Specific Threshold 3 • Continued exceedance of Threshold 1 for 3 years	**Notification** • Minto Management, SFN, YG Inspector • Include in scheduled Water Use Licence reporting **Review** • All review measures for ST2 **Evaluation** • All evaluation measures for ST2 **Action** • If all results are confirmed, then non-degradation is declared unattainable, and the specific indicator (i.e. parameter) is then subject to the AMP framework in Table 2-7 (50%AC-WQOs).

In the case of the non-degradation condition being declared unattainable for any given specific indicator through the decision framework above, that parameter will be subject only to the following AMP framework which applies the 50% of assimilative capacity water quality objectives. Until such time, the specific indicators will be subject to both frameworks to ensure that actions are implemented in the case where 50% assimilative capacity WQOs are approached or exceeded early in the post closure period. The definition of attainment, of the closure WQOs are detailed in the RCP 2018-01 and is used as the basis for Specific Threshold 4 in Table 2-9, below. Specifically, as outlined in greater detail in the RCP, attainment is considered to be achieved if Specific Threshold 4 is not triggered.

Table 2-9: Specific Indicators, Performance Thresholds and Responses for Surface Water Quality at Station W2 in Minto Creek, using the 50% Assimilative Capacity Water Quality Objectives.

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
Aqueous Concentrations at Station W2 for parameters with water quality objectives (See Table **2-7** above for specific threshold values)	**Specific Threshold 1** • Exceedance of SPT1 from Table 2-7	**PCII**	**Notification** • Minto Management • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • Compare with W3/CWTS results **Action** • On next scheduled sampling event, ensure sampling conducted at mine site stations W12 and W45 (open pits) if not already scheduled.
	Specific Threshold 2 • Exceedance of SPT1 in **2 consecutive samples at W2** where evaluation confirmed mine loading responsible for first exceedance	**PCII**	**Notification** • Minto Management • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • Compare with W3 results. If comparison with W3 result indicates that mine loadings are responsible for exceedance; and validation confirms original result, then: o Evaluate causes for load contributions and develop investigation plan; and o A trend analysis will be conducted by a qualified person. **Action** • Implement investigation plan, including at a minimum: o Re-sampling both W2 and W3 within **one month** of original sample result review; and o Site investigation of candidate load contributions. • Review results of investigation and prepare recommendations if appropriate. • If trend analysis suggests WQO exceedance within one year, then initiate actions for Specific Threshold 3.
	Specific Threshold 3 • 50IDPE-WQO exceeded at W2 in a single sample where evaluation confirms that mine loading is responsible for exceedance.	**PCII**	**Notification** • Minto Management, SFN, YG Inspector • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol o Review laboratory QA/QC report o Verify original result, or re-run sample if laboratory error indicated **Evaluation** • Compare with W3 results. If comparison with W3 result indicates mine loadings responsible for exceedance – and verification confirms original result – then: o Evaluate candidate causes for load contributions and develop investigation plan (or review/revise as appropriate) **Action** • Implement investigation plan, including any reviews/revisions, and at a minimum: o Re-sampling both W2 and W3 within **two weeks** of original sample result review; and o Site investigation of candidate load contributions • Review results of investigation and prepare recommendations if not already prepared from ST2. • Implement recommendations from above.
	Specific Threshold 4 • Exceedance of 50IDPE-WQO in **2 consecutive samples (scheduled or re-sample) at W2**	**PCII**	**Notification** • Minto Management, SFN, YG Inspector • Include in scheduled Water Use Licence reporting **Review**

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
	or twice in one year where evaluation confirmed mine loading responsible for first exceedance **Or** • 50CT-WQO exceeded at W2 by a 3 year rolling average value (ST-4 = Attainment as defined in the RCP 2018-01)		• Compare with W3 results • Follow QA/QC investigative protocol **Evaluation** • Compare with W3 results. If comparison with W3 result indicates mine loadings responsible for exceedance; and verification confirms original result, then: o Evaluate candidate causes for load contributions and develop investigation plan (or review/revise as appropriate) o Provide investigation plan to SFN/YG Inspector • Engage a qualified individual for the evaluation of potential effects to aquatic resources (i.e. Compare to Predicted No Effect Concentration (PNEC) and apply Biotic Ligand Model (BLM) for D-Cu, invertebrate tissue for Se, compare with calculated acute guideline for D-Cd.) This may include but not be limited to the evaluation of existing data and results from effluent toxicity, surface water toxicity, sediment chemistry, sediment toxicity, benthic invertebrate community structure and fish health testing/programs. **Action** • Implement monthly monitoring at W2 and W3/CWTS if recommended by qualified individual evaluating for the potential for effects at aquatic resources. • Revise investigation plan, including at a minimum: o Re-sampling both W2 and W3 within **one week** of original sample result review; and o Additional site investigation of load contributions • Review results of investigation and prepare additional recommendations • Implement recommendations arising from investigations.
	Specific Threshold 5 • 50IDPE-WQO exceeded in **3 consecutive samples (scheduled or re-sample) at W2** where evaluation confirmed mine loading responsible for exceedances	**PCII**	**Notification** • Minto Management, SFN, YG Inspector • Include in Water Use Licence reporting **Review** • Compare with W3/CWTS results and any other WQ data as appropriate • Follow QA/QC investigative protocol **Evaluation** • All evaluation measures for ST4 **Action** • Maintain monthly monitoring and collect samples at greater frequency as required. • Review and revise mitigation plans with qualified individuals, and Implement necessary reasonable and practical measures to reduce contaminant loading from mine to Minto Creek. This could include consideration of: additional passive treatment installations. • Actions will continue until performance thresholds are no longer exceeded.

2.1.4 McGinty Creek

2.1.4.1 Description

The Minto North deposit is an extension of the mineralized corridor being mined within the Minto Creek catchment. It is within the McGinty Creek catchment area, to the north of Minto Creek. Minto has been monitoring surface water quality in the McGinty Creek catchment since 2009, and the results from monitoring program (until the end of 2015) are presented in McGinty Creek Water Quality Characterization, July 2016 (AEG, 2016).

As described in the Water Management Plan (Minto Explorations Ltd., 2015), during the recently completed active mining of Minto North, Minto actively managed water directly impacted by open pit mining with pump trucks. With surface mining in Minto North now complete, there is no active water management, with runoff and meteoric water now allowed to accumulate in the pit.

The AMP framework below provides a decision-based structure with the goal of avoiding changes to background water quality in lower McGinty Creek that result from completed mining activities at Minto North. Station MN-4.5 is the monitoring station on the lower main stem of McGinty Creek, after the north and south tributaries converge, and near the confluence of McGinty Creek with the Yukon River. Surface water quality at the MN-4.5 monitoring point is subject to periodic TSS influences from catchment area that are beyond Minto's control, so dissolved metals concentrations will be used to track influences from the Minto North development.

2.1.4.2 Risk Narrative

Increase in contaminant concentrations from completed Minto North mining activities causes unacceptable changes to surface water quality in McGinty Creek.

2.1.4.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to water quality and the monitoring program are provided below in Table 2-11. The specific indicators are total suspended solids and contaminant (nitrogen species and dissolved metals/metalloids) concentrations for parameters identified in the Water Use Licence QZ14-031 Table 2 – Water Quality Objectives. Thresholds are all based on a proposed water quality objective (WQO), and the specific indicators station, as identified above, is MN-4.5. The discussion below applies to data collected from this station. The selection of this indicator station, along with other aspects of the AMP framework, is consistent with the approach taken for closure water quality objective development in Minto Creek (i.e. downstream indicator station, non-degradation water quality objectives, and a focus on dissolved metal concentrations.)

Essentially, Minto has adopted the statistical definitions of non-degradation (from the discussions and agreement with SFN regarding Minto Creek closure water quality objectives) as the basis of developing these revised McGinty Creek Water Quality Objectives. Monthly monitoring has continued in the McGinty Creek catchment, and the background dataset has been updated to include all monitoring data from initiation of the program in May 2009 until July 2015 (stripping of the Minto North Pit began in August 2015.) Similar to the data treatment used in Minto Creek, data from monitoring stations were collapsed into monthly results (most monitoring has been undertaken monthly anyway, and this was only required for May 2009 when sampling was weekly.) All monthly

data were then used to calculate the 95th percentile, for use as the maximum water quality objective, or individual data point evaluator (IDPE). The monthly data were then also grouped by year, and the annual medians were calculated. The 95th percentile of these annual medians was calculated for each station to generate the central tendency evaluator (CTE).

These objectives form the basis of the AMP thresholds. Values lower than the WQOs have been selected as early warning thresholds (ST 1) prior to the WQOs themselves forming the higher level specific thresholds (ST2 and ST3). For these thresholds, the 85th percentile of the same data selected. Utilizing a statistic such as this is more effective and reliable than using a percentage (e.g. 75%) of maximum value, as it considers the actual statistical distribution of the background data.

The actual calculated threshold values are presented in Table 2-10.

The monitoring results that are evaluated and utilized for this component of the AMP are a requirement of the Surface Water Surveillance Program of the EMSRP (Minto Explorations Ltd., 2016). The monitoring data will be compared to the specific performance thresholds monthly (by the end of the month following the month in which samples were collected) - this corresponds to the existing monthly reporting schedule.

Table 2-10: Specific Thresholds for McGinty Creek, Station MN-4.5

Analytes	Specific Threshold 1 (85th percentile)	Water Quality Objectives (Specific Thresholds 2 and 3)	
		Individual Data Point Evaluator (95^{th} percentile)	Central Tendency Evaluator (95th percentile of annual medians)
Total Suspended Solids (mg/L)	52.0	269	32.0
Ammonia (mg/L)	0.040	0.12	0.046
Nitrite (mg/L)	0.006	0.05	0.005
Nitrate (mg/L)	0.200	0.232	0.083
Dissolved Aluminum (µg/L)	48.6	135.0	47.0
Dissolved Arsenic (µg/L)	0.55	0.61	0.54
Dissolved Cadmium (µg/L)	0.026	0.041	0.015
Dissolved Chromium (µg/L)	1.0	1.0	1.0
Dissolved Copper (µg/L)	2.9	3.5	2.8
Dissolved Iron (µg/L)	334	403	358
Dissolved Lead (µg/L)	0.20	0.20	0.20
Dissolved Molybdenum (µg/L)	1.0	1.0	1.0
Dissolved Nickel (µg/L)	1.6	1.8	1.6
Dissolved Selenium (µg/L)	0.17	0.20	0.16
Dissolved Silver (µg/L)	0.020	0.020	0.020
Dissolved Zinc (µg/L)	5.0	5.2	5.0

Table 2-11: Specific Indicators, Performance Thresholds and Responses for Surface Water Quality in McGinty Creek

Specific Indicators	Specific Performance Thresholds	Specific Responses
Aqueous Concentrations at Station MN-4.5 for parameters with Water Quality Objectives. (See Table 2-10 above for Water Quality Objective and threshold values.)	**Specific Threshold 1** • Exceedance of ST1 value in two consecutive samples **(scheduled or re-sample)**	**Notification** • Minto Management • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • Compare with MN-1.5 results • A trend analysis will be conducted by Minto's senior level environmental personnel. **Action** • If comparison with other results suggests that Minto North loading may be responsible for exceedance and validation confirms original result, then: o Evaluate causes for load contributions, and o If trend analysis suggests WQO exceedance within one year, then initiate actions for threshold 2.
	Specific Threshold 2 • Exceedance of the IDPE-WQO (grab vs. IDPE, or 12-month moving average vs. CTE)	**Notification** • Minto Management • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • Measures from ST1, and • Engage a qualified professional to evaluate potential effects to aquatic resources • A trend analysis will be conducted by Minto's senior level environmental personnel. **Action** • If comparison with results suggests that Minto North loadings are responsible for exceedance and validation confirms original result, then: o Re-sample MN-1.5 and MN-4.5 within **two weeks** of original sample result review; and o Evaluate causes for load contributions o Develop a mitigation strategy with recommendations based on the findings of the potential effects to aquatic resources evaluation • Actions will continue until performance thresholds are no longer exceeded.
	Specific Threshold 3 • Exceedance of the IDPE-WQO (as defined above) in **2 consecutive samples (scheduled or re-sample)** where evaluation confirmed mine loading responsible for first exceedance	**Notification** • Notify management, SFN and YG • include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated **Evaluation** • Measures from ST2 **Action** • If comparison with results (evaluation, above) suggests that Minto North loadings are STILL responsible for exceedance; and validation confirms original result, then: o If not already implemented, increase monitoring frequency. o Implement recommendations from mitigation strategy (threshold 2 response). This could include batch water treatment in the Minto North pit if determined feasible and appropriate. o Actions will continue until performance thresholds are no longer exceeded.

2.2 Groundwater Quality

2.2.1 Minto Creek Watershed

2.2.1.1 Description

Groundwater quality has the potential to be important in terms of contributions to surface water quality. Groundwater contributes to streamflow as baseflow, which is typically most important during the autumn/winter low flow season when surface water flows are minimal.

All mine workings and waste facilities within in the Minto Creek catchment are located upgradient (and west) of the Water Storage Pond (WSP). During final closure the WSP will be decommissioned and a passive treatment system (wetlands) established within its footprint.

The groundwater monitoring supporting the Closure AMP framework is defined and described in the Monitoring and Maintenance program, which is included in Section 7.11 of the RCP. The monitoring requirements build on those of the Groundwater Monitoring Program of the Environmental Monitoring, Surveillance and Reporting Plan (EMSRP) (Minto Explorations Ltd., 2018b) implemented in the mine operations phase. The purpose of the operational groundwater monitoring program is to provide information to better understand the potential for off-site migration of contamination and to better understand and define source terms and conditions of waste emplacement, for comparison to source terms used in the predictive Water Balance and Water Quality model. An additional objective of the operational groundwater monitoring plan is to provide for the development of baseline hydrogeological conditions in areas where future mine components are being planned.

Groundwater monitoring is carried out both upgradient and downgradient of the WSP, via multi-level monitoring wells at MW12-06, MW12-05 and MW17-12 (MW12-05 is located downgradient of the new MW17-12. Both wells will be monitored until comparison by a qualified professional determines that the use of MW12-05 should be discontinued).

As described in the 2018 Minto Groundwater Characterization Hydrogeologic Conceptual Model Update Report (SRK 2018), groundwater coming from the mine area (the western and highest elevation portion of the Minto Creek catchment) is expected to discharge to surface water in the vicinity of the WSP. Minimal groundwater from the mine area is expected to discharge to Minto Creek down gradient of the WSP. Surface water downgradient of the WSP is monitored at several stations including station W2 located near the Yukon River. W2 is 6 km downstream of the Minto lease boundary and water reporting to W2 is subject to influences from groundwater and surface water outside the mine area.

The Closure AMP framework below complements the RCP water management plan for the mine site with a decision-based structure for ensuring that negative impacts to lower Minto Creek from mining and closure activities are avoided.

2.2.1.2 Risk Narrative

Flux of geochemical load from a mine component via groundwater pathways causes surface water quality objectives to be in exceeded in Minto Creek at station W2.

2.2.1.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to groundwater quality in Minto Creek watershed are provided in Table 2-12. Specific Performance Thresholds (SPTs) are defined for each of the Effluent Standards (ES) parameters identified in Clause 9 (a), Table 1 of Water Use Licence QZ14-031, with the exception of pH, oil and grease, iron and nitrite (as discussed later in this section). Additionally, although it is not specified in Clause 9(a), sulphate has been included in the SPTs. Table 2-13 compiles the concentration of background groundwater for each Specific Indicator and the Specific Performance Threshold values.

The concentrations of the background groundwater are based on the respective median parameter concentrations at monitoring well MW09-03 (including all monitoring ports). At present (February 2018), the MW09-03 monitoring record is still considered to be the most representative indicator of the baseline groundwater conditions for the entire project site. MW09-03 was installed in 2009 and monitored to collect baseline data downgradient of the Minto North ore body- this well is located near the southern limit of the McGinty Creek catchment close to the surface water divide with Minto Creek. Pre-mining groundwater concentrations for certain parameters such as cadmium and iron were relatively high, likely due to the adjacent highly mineralized zone. Mining of this ore body began in Q3 2015 and as such the monitoring record prior for the 2009-2015 period reflects baseline conditions. MW17-08 was installed in late 2017 to allow groundwater quality monitoring within the Minto Creek catchment upgradient of any mine disturbances. Once an adequate record of the groundwater quality from MW17-08 is developed, the groundwater quality at MW09-03 and MW17-08 will be compared by a qualified professional to verify and assess if background groundwater concentrations used in the current operational AMP remain appropriate. Minto expects that an adequate record will be available after 3 years of monitoring. In addition, as reported in SRK (2018) elevated sulphate at MW12-05 is believed to not be a result of mine impacts, but represents elevated background concentrations. Sulphate will continue to be reviewed to improve confidence in this finding, and other parameter will be reviewed within this context. Other new wells installed in 2017 will be monitored in accordance with Minto's EMSRP, but will not be included in the AMP.

Iron and nitrite have been excluded from the Specific Indicators for groundwater quality because of the magnitude of the natural variability observed in groundwater at Minto. In addition, pH as also been excluded as a Specific Indicator for groundwater quality because it is not as useful an early warning indicator as sulphate and metal/ metalloid concentrations, and Oil & Grease has been excluded because it is not relevant as a specific indicator for groundwater. Although iron, nitrite and pH are excluded from the Specific Indicator list, these three constituents are monitored and would be included in the review of groundwater quality if an SPT was exceeded.

Three SPTs have been defined for the Minto Creek watershed. Rationale for development of the SPTs is as follows:

- SPT-1 corresponds to a trend-based assessment designed to flag a potential rapid increase in groundwater loadings that has not yet exceeded concentration-based thresholds. The assessment is structured to determine if an indicator has increased significantly compared to the last sampling event. The assessment will be performed as followed:

  $$((C_n - C_{n-1})/(C_{SPT-2} – C_n) > 0.2$$

  Where:

  - C_n = the parameter concentration of groundwater from the latest sampling event;
  - C_{n-1} = the parameter concentration of groundwater from the last sampling event ;

- C_{SPT-2} = the parameter concentration for the SPT-2;

The SPT-1 provides a conservative threshold considering that it is weighted against the concentration for the SPT-2. The SPT-1 will be increasingly sensitive to change in concentrations between two sampling events as groundwater approaches the SPT-2, since the size of the denominator decreases as the SPT-2 value is approached.

- SPT-2 generally corresponds to the EQS concentrations defined in Clause 9 (a), Table 1 of Water Use License QZ14-031 (with four exceptions (Cr-D, Cu-D, Ni-D and sulphate) as indicated in the notes to Table 2-13). The EQS are defined for surface water discharge, and the mine is not permitted to discharge water that exceeds any EQS guidelines to surface water. As SPT-2 applies to groundwater (not surface water) concentrations in single zone, this is a highly conservative threshold.
- SPT-3 generally corresponds to the estimated concentrations in groundwater that would be necessary to cause exceedance of the Water Quality Objectives in lower Minto Creek, at W2, under long term steady state conditions (exceptions are listed in the notes to Table 2-13). These concentrations were determined by conducting a mass loading calculation to determine the groundwater concentrations that would be necessary to cause exceedance of WQOs in lower Minto Creek during low flow periods where all streamflow is derived from groundwater discharge. The mass loading calculation was structured to represent a low-flow period when all surface flows in lower Minto Creek originate from groundwater discharge to the creek. The calculation was done using the following formula:

$$C_{gw\text{-}mine_max} = ((Q_{W2} \times C_{W2\text{-}WQO}) - (Q_{gw\text{-}bgrnd} \times C_{gw\text{-}bgrnd}))/Q_{gw\text{-}mine}$$

Where:

- $C_{gw\text{-}mine_max}$ = the indicator parameter concentration of all groundwater upgradient of MW12-05 that reports to Minto Creek that would be required to cause surface water to exceed the WQO at station W2.
- Q_{W2} = the combined groundwater flow discharging to Minto Creek (total groundwater discharge to Minto Creek from the 2015 groundwater model update) during low flow periods
- $C_{W2\text{-}WQO}$ = the WQO parameter concentration for surface water at W2
- $Q_{gw\text{-}bgrnd}$ = background groundwater flow discharging to Minto Creek (groundwater discharge down gradient of the Water Storage Pond from the 2015 groundwater model update) during low flow periods
- $C_{gw\text{-}bgrnd}$ = background groundwater concentration (based on median concentrations from the 2009-2015 baseline monitoring period in MW09-03)
- $Q_{gw\text{-}mine}$ = the estimated groundwater flow from the mine at the Water Storage Pond (from the 2015 groundwater model update)

If groundwater concentration of one indicator reached the SPT-3 in an individual port, the quality of the lower Minto Creek at W2 would not yet exceed the Water Quality Objective due to the contribution of groundwater from un-impacted areas of the Minto Creek watershed and the fact that groundwater concentrations at a specific monitoring well port represents only a portion of the flow and not the whole groundwater flow field. The SPT-3

provides therefore a conservative threshold for action before any significant effect would be observed in surface water.

The monitoring results that are evaluated and utilized for this component of the AMP are a requirement of the Groundwater Monitoring Program of the EMSRP (Minto Explorations Ltd., 2018b). The monitoring data will be compared to the specific performance thresholds quarterly.

Examples of actions that may arise from recommendations include:

- Continuation of monitoring;
- Continuation monitoring with an increase in monitoring frequency;
- Development of additional monitoring points and monitoring of those newly-established monitoring locations;
- Completion of appropriate risk assessment;
- Development and execution of a focused study to better understand the cause of exceedance.

Table 2-12: Specific Indicators, Performance Thresholds and Responses for Groundwater Quality in Minto Creek

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
Aqueous concentrations in samples collected from multi-level groundwater monitoring wells MW12-05 or MW12-06 for the following parameters with water quality objectives: • Dissolved Aluminum • Dissolved Arsenic • Dissolved Cadmium • Dissolved Chromium • Dissolved Copper • Dissolved Iron • Dissolved Lead • Dissolved Molybdenum • Dissolved Nickel • Dissolved Silver • Dissolved Selenium • Dissolved Zinc • NH_4-N • NO_3-N • Sulphate	**SPT-1** • Exceedance of SPT-1 for any sample collected during routine monitoring from multilevel groundwater monitoring wells MW12-05 or MW12-06.	**All Stages**	**Notification** • Minto Management • Include in scheduled Water Use Licence monthly reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated o Timing: initiate within 1 week of triggering SPT **Evaluation** • Review of groundwater monitoring data (trend analysis included) to be undertaken by qualified professional, and appropriate recommendations to be developed o Review must consider the risk narrative (i.e. exceedance of surface water quality objectives as a result of groundwater flux) • Submit recommendations to regulator for review and approval o Timing: submit within 1 week of receipt of recommendations **Action** • Follow recommendations arising from review undertaken by qualified professional. • Include the trend analysis in scheduled reporting o Timing: initiate implementation of recommendations within 1 month of receipt of approval from regulator
	SPT-2 • Exceedance of SPT-2 concentrations in 2 consecutive samples (scheduled or re-sampled) collected during routine monitoring from multilevel groundwater monitoring wells MW12-05 or MW12-06.	**All Stages**	**Notification** • Minto Management • Include in scheduled Water Use Licence monthly reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated o Timing: initiate within 1 week of triggering SPT **Evaluation** • Review of groundwater monitoring data (including trend analysis) to be undertaken by qualified professional, and appropriate recommendations to be developed o Review must consider the risk narrative (i.e. exceedance of surface water quality objectives as a result of groundwater flux) o Timing: initiate within 1 week of QA/QC review validating original results • Submit recommendations to regulator for review and approval o Timing: submit within 1 week of receipt of recommendations **Action** • Follow recommendations arising from review undertaken by qualified professional. o Timing: initiate implementation of recommendations within 1 month of receipt of approval from regulator

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
	SPT-3 • Exceedance of SPT-3 in 2 consecutive samples (Scheduled or re-sampled) collected during routine monitoring from multilevel groundwater monitoring wells MW12-05 or MW12-06.	**All Stages**	**Notification** • Minto Management • Include in scheduled Water Use Licence monthly reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original result, or re-run sample if a laboratory error is indicated o Timing: initiate within 1 week of triggering SPT **Evaluation** • Review of groundwater monitoring data (including a trend analysis) to be undertaken by qualified professional, and appropriate recommendations to be developed o Review must consider the risk narrative (i.e. exceedance of surface water quality objectives as a result of groundwater flux) o Timing: initiate within 1 week of QA/QC review validating original results • Submit recommendations to regulator for review and approval o Timing: submit within 1 week of receipt of recommendations **Action** • Follow recommendations arising from review undertaken by qualified professional. o Timing: initiate implementation of recommendations within 1 month of receipt of approval from regulator • Increase monitoring frequency to monthly sampling of all monitored zones in the affected multi-level well for a period to be defined by qualified professional.

Table 2-13: Background concentrations and SPTs for the groundwater monitoring in the Minto Creek watershed

Minto Creek	***Concentrations (mg/L)***													
	Ag-D	Al-D	As-D	Cd-D	Cr-D	Cu-D	Pb-D	Mo-D	Ni-D	Se-D	Zn-D	N-NO3	Ammonia	Sulphate[2]
SPT-3 [1]	0.00046	0.48	0.025	0.0060	0.0030	0.060	0.020	0.34	0.33	0.0093	0.13	45	1.0	4951
SPT-2	0.00030	0.30	0.015	0.0030	0.0015[4]	0.030 [4]	0.012	0.22	0.165[4]	0.0060	0.090	27	0.75	1000
Background Groundwater[3]	0.00001	0.0045	0.0001	0.00002	0.00050	0.0014	0.0001	0.006	0.001	0.0002	0.006	0.07	0.051	12

Notes:

1: For most Specific Indicators, SPT-3 is the calculated concentration that all groundwater from the mine catchment must attain to reach the Operational Water Quality Objectives at W2. Model flows were based on 2015 Groundwater Model Update. 15 L/s is the total groundwater discharging to Minto Creek. 3 L/s is the estimated groundwater flow from the Minto Creek catchment up gradient of the Water Storage Pond. The exceptions are dissolved chromium, dissolved copper and dissolved nickel, for which SPT-3 is equal to the Effluent Quality Standard value that applies to surface water discharge from the mine site.

2: There is no Effluent Quality Standard for sulphate. For the SPT-2, the guideline for Aquatic Life from the Contaminated Site Regulation Schedule 3 was used as a replacement.

3: The background concentration in groundwater is calculated as the median of concentrations observed at groundwater monitoring well MW09-03.

4: SPT-2 for dissolved copper, dissolved chromium and dissolved nickel set at one-half (50%) of Effluent Quality Standard (SPT-2 for all remaining parameters (other than sulphate) is equal to (100% of) the Effluent Quality Standard).

2.2.2 McGinty Creek Watershed

2.2.2.1 Description

The mine workings in the McGinty Creek catchment are limited to the Minto North Pit. The catchment area of the Minto North Pit is roughly 15 ha and the catchment area of the McGinty Creek watershed is roughly 3400 ha (SRK 2013); in other words, the Minto North Pit catchment is roughly 0.4% of the total McGinty Creek catchment area. While a groundwater model encompassing the full extent of the McGinty Creek catchment has not been developed, it is clear that only a very small proportion of the groundwater in the McGinty Creek watershed can be affected by the Minto North Pit.

The monitoring supporting the closure AMP framework is defined and described in the Post-Closure Monitoring Plan of the RCP. The monitoring requirements build on those of the operational Groundwater Monitoring Plan (part of the EMSRP ((Minto Explorations Ltd., 2018b)). Groundwater in the McGinty Creek catchment downgradient of the Minto North Pit is currently monitored at multi-level monitoring well MW09-03. MW09-03 was installed in 2009 and monitored to collect baseline at the Minto North ore body. Mining of this ore body began in Q3 2015 and as such the monitoring record prior for the 2009-2015 period reflects baseline conditions. An additional multi-level groundwater well MW17-11 was installed downgradient of the Minto North Pit in late 2017 to fulfill WUL requirements.

2.2.2.2 Risk Narrative

Flux of geochemical load from the Minto North Pit via groundwater pathways causes surface water quality objectives to be in exceeded in McGinty Creek at station MN4.5.

2.2.2.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to groundwater quality in McGinty Creek watershed are provided below in Table 2-14. SPTs are defined for each of the Effluent Standards (ES) parameters identified in Clause 9 (a), Table 1 of Water Use License QZ14-031, with the exception of pH, Oil & Grease, iron and nitrite (for reasons described in the Minto Creek groundwater section (Section 2.2.1). Additionally, although it is not specified in Clause 9(a), sulphate has been included in the SPTs. The concentration of background groundwater and the Specific Performance Threshold values for each Specific Indicator and each zone (i.e. individual well port) of MW09-03 are compiled in Table 2-15. The monitoring record for MW09-03 for the 2009-2015 period reflects baseline conditions (prior to the development of the open pit). Pre-mining groundwater concentrations for certain parameters such as cadmium and iron have been relatively high, likely due to the adjacent highly mineralized zone.

Two Specific Performance Thresholds have been defined for McGinty Creek watershed groundwater- both are conservative given the application of the thresholds at individual ports, the small proportion of catchment groundwater that will be influenced by the mine workings and the expected slow rates of groundwater movement. Rationale for development of the two specific performance thresholds is as follows:

- SPT-1: Three consecutive exceedances of the 75th percentile background level in a single monitoring port.
    - The specification of three consecutive exceedances is intended to avoid triggering the AMP unnecessarily, but to ensure that any sustained increase from baseline conditions receives appropriate scrutiny.
- SPT-2: Three consecutive exceedances of the 95th percentile background level in a single monitoring port.

- The specification of three consecutive exceedances is intended to avoid triggering the AMP unnecessarily, but to ensure that any sustained increase from baseline conditions receives appropriate scrutiny.

The monitoring results that are evaluated and utilized for this component of the AMP are a requirement of the operational Groundwater Monitoring Program of the EMSRP (Minto Explorations Ltd., 2018b) and the Post-Closure Monitoring Plan of the RCP.

Table 2-14: Specific Indicators, Performance Thresholds and Responses for Groundwater Quality in McGinty Creek

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
Aqueous concentrations in samples collected from multi-level groundwater monitoring well MW09-03 for the following parameters: • Dissolved Aluminum • Dissolved Arsenic • Dissolved Cadmium • Dissolved Chromium • Dissolved Copper • Dissolved Lead • Dissolved Molybdenum • Dissolved Nickel • Dissolved Silver • Dissolved Selenium • Dissolved Zinc • NH_4-N • NO_3-N • Sulphate	**Specific Threshold 1** • Three consecutive exceedances of the 75th percentile value* from the baseline period (2009-2015) in routine monitoring results from a single monitoring port in MW09-03. *values provided in Table 2-13	**All Stages**	**Notification** • Mine/Project Manager • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original results, or re-run samples if a laboratory error is indicated **Evaluation** • Review of groundwater monitoring data to be undertaken by qualified professional, and appropriate recommendations to be developed o Review must consider the risk narrative (i.e. exceedance of surface water quality objectives as a result of groundwater flux) **Action** • Follow recommendations arising from review undertaken by qualified professional.
	Specific Threshold 2 • Three consecutive exceedances of the 95th percentile value* from the baseline period (2009-2015) in routine monitoring results from a single monitoring port in MW09-03. *values provided in Table 2-13	**All Stages**	**Notification** • Mine/Project Manager • Include in scheduled Water Use Licence reporting **Review** • Follow QA/QC investigative protocol: o Review laboratory QA/QC report o Validate original results, or re-run samples if a laboratory error is indicated **Evaluation** • Review of groundwater monitoring data to be undertaken by a qualified professional, and appropriate recommendations to be developed o Review must consider the risk narrative (i.e. exceedance of surface water quality objectives as a result of groundwater flux) **Action** • Follow recommendations arising from review undertaken by qualified professional. • Increase monitoring frequency of all monitored zones in the affected multi-level well for a period to be defined by qualified professional.

Table 2-15: Background Concentrations and SPTs for the Groundwater Monitoring in the McGinty Creek Watershed

Minto Creek		*Concentrations (mg/L)*[1]													
		Ag-D[3]	Al-D	As-D	Cd-D	Cr-D[3]	Cu-D	Pb-D[3]	Mo-D	Ni-D	Se-D	Zn-D	N-NO3	NH_4-N	Sulphate
MW09-03-01	95th Percentile	0.000016	0.0101	0.00081	0.000272	0.000575	0.00625	0.0003	0.026	0.0060	0.0015	0.022	0.28	0.133	38
	75th Percentile	0.000010	0.0065	0.00011	0.000075	0.000500	0.00155	0.0001	0.005	0.0021	0.0001	0.013	0.13	0.067	24
	Median[2]	0.000010	0.0045	0.00005	0.000022	0.000500	0.00031	0.0001	0.004	0.0015	0.0001	0.006	0.07	0.045	22
MW09-03-02	95th Percentile	0.000034	0.0095	0.00092	0.000272	0.000796	0.01080	0.0002	0.062	0.0026	0.0040	0.015	0.07	0.282	67
	75th Percentile	0.000018	0.0073	0.00074	0.000031	0.000500	0.00263	0.0001	0.018	0.0009	0.0005	0.010	0.03	0.230	7
	Median[2]	0.000010	0.0062	0.00067	0.000026	0.000500	0.00122	0.0001	0.017	0.0005	0.0002	0.008	0.01	0.210	1
MW09-03-03	95th Percentile	0.000010	0.0075	0.00014	0.000069	0.000500	0.00500	0.0003	0.018	0.0011	0.0004	0.011	0.54	0.058	13
	75th Percentile	0.000010	0.0047	0.00005	0.000023	0.000500	0.00247	0.0001	0.006	0.0005	0.0004	0.008	0.50	0.020	12
	Median[2]	0.000010	0.0025	0.00005	0.000015	0.000500	0.00174	0.0001	0.005	0.0005	0.0003	0.003	0.48	0.012	11

Notes:

1: For monitoring results where concentrations were below the analytical detection limits, a concentration of half the detection limit was adopted for calculation purposes.

2: For AMP purposes, the background concentration in groundwater at MW09-03 is defined as the median concentration observed in each port over the 2009-2015 baseline monitoring period.

3: For Ag-D, Cr-D and Pb-D, most 2009-2015 concentrations were at the limit of analytical detection, and as such the calculated 75th and 95th percentile values are skewed low.

2.3 Water Management Infrastructure

2.3.1 Description

The closure water conveyance infrastructure is a network of channels, erosion protection features, and energy dissipation structures that convey surface water from the mine surface to Minto Creek. The network of channels is designed to collect, intercept, and convey water through the mine in its closed state. In addition to the conveyance system, three reservoirs will be present into active closure and post closure phases. Two of the reservoirs, the Main Pit Lake and the Area 2 Pit Lake will remain in perpetuity, while the third one, the WSP, will be decommissioned and the dam deconstructed. The Water Storage Dam has been excluded from the closure AMP as the facility is to be managed according to its Operation, Maintenance and Surveillance (OMS) Manual (Tetra Tech EBA, 2014) until it is decommissioned. A Constructed Wetland Treatment System (CWTS) is to be constructed within the footprint currently occupied by the WSP as part of mine closure works at Minto Mine.

Conveyance and collection structures and reservoirs are used to convey drainage from the major waste infrastructure to treatment systems prior to release. As discussed in the Section 7.7.1 of the RCP the water treatment plant used during operations will remain in operational until WUL criteria can be achieved consistently without its utilization.

The design criteria for the water conveyance system is described in the Section 3 of the Minto Mine – Closure Water Conveyance System Design Update Report (Appendix G2, Minto Mine RCP). Design details specific to each of the five conveyance channels are provided in Section 5 of the design report.

The water management system is able to deal with most foreseeable conditions that may be encountered though the mine closure. However, certain unforeseen conditions may require an adaptive response as described below.

For clarity, this AMP component applies to the following water management infrastructure:

- Water Storage Pond and WSP Dam;
- Tailings Diversion Ditch;
- All primary and secondary conveyance ditches and pit inflow/outflow structures;
- CWTS head ponds;
- CWTS High Flow Bypass Ditch;

2.3.2 CWTS treatment cells and outflow channel.Risk Narrative

A mass failure of one of the conveyance and collection facilities has the potential to endanger the health and safety of site employees or visitors, or lead to an increase in contaminant loadings from the mine and subsequent adverse effects to aquatic resources in the receiving environment (lower Minto Creek).

2.3.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to water management are provided below in Table 2-16. The monitoring results that are evaluated and utilized for this component of the AMP are a requirement of the Post-Closure Monitoring Program.

Table 2-16: Specific Indicators, Performance Thresholds and Responses for Water Management Infrastructure

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses

• Failure or blockage of conveyance structures resulting in changed flow paths and/or overtopping of drainage ditches
• Failure of upstream diversions leads to excessive infiltration into upgradient base of DSTF, resulting in increased flows and/or metal loads and unacceptable water quality conditions downstream.
• Occurrence of a major precipitation event
• Increased sediment load in ditch flow and or downstream (increased TSS)
• Pit wall failure
	Specific Threshold 1
	• Observation of unusual occurrence including:
	  • Partial blockage;
	  • Physical damage;
	  • Turbulent flow where laminar flow is expected or is normal;
	  • Cracks, settlement of sloughing;
	  • Downstream movement of channel bedding material (rip rap or liner);
	  • Erosion channels or flowing water outside of the designed structures;
	  • Abnormal seepage;
	  • Rainfall event that exceeds the design criteria;
	  • Increase turbidity of flow.
		All Stages	**Notification**
			• Mine/Project Manager
			• Geotechnical Engineer
			• Include in scheduled WUL reporting
			Review
			• Review previous inspection reports, survey data and precipitation records
			Evaluation
			• Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of the need for design modifications
			Action
			• Inspect area to determine source of blockage and potential for repeated occurrence
			• Develop plan to address the need for repairs or modifications.
			• Increase frequency of monitoring as recommended by a qualified professional

	Specific Threshold 2
	• Observation of unusual occurrence including:
	  • Complete blockage;
	  • Substantial physical damage;
	  • overtopping of drainage ditches;
	  • Downstream water quality exceeds TSS criteria.
		All Stages	**Notification**
			• Mine/Project Manager
			• Geotechnical Engineer
			• Include in scheduled WUL reporting
			Review
			• Review previous inspection reports, survey data and precipitation records.
			Evaluation
			• Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of the need for design modifications
			Action
			• Inspect area to determine source of blockage and potential for repeated occurrence
			• Immediately develop and implement a plan to address the need for repairs or modifications
			• Increase frequency of monitoring as recommended by a qualified professional

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
	Specific Threshold 3 • Complete compromise of functionality.	**All Stages**	**Notification** • Mine/Project Manager, SFN and YG • Geotechnical Engineer • Include in scheduled WUL reporting **Review** • Review previous inspection reports, survey data and precipitation records. **Evaluation** • Engage a qualified professional to evaluate potential effects to aquatic resources. **Action** • Inspect the area for any other signs of instability. • Complete a ground survey of the area accommodate modifications/repairs. • Follow any additional recommendations of the Geotechnical Engineer. At a minimum, the Engineer will consider the need for modifications to the conveyance and retention structures and the need for an increased frequency of monitoring.

2.4 Physical Stability of Waste Rock, Overburden and Tailings Storage Facilities

2.4.1 Description

The physical stability of the site, during Post Closure I and II periods, will be monitored according to the Monitoring and Maintenance section (7.11) of the Minto Mine RCP. During active closure the physical stability of the waste rock, overburden dumps, tailings and water storage facilities are will be monitored according to the Physical Monitoring Plan, which forms part of the Environmental Monitoring, Surveillance and Reporting Plan (Minto Explorations Ltd., 2016). These monitoring plans describe the inspection and instrumentation data collection frequencies, instrument locations, installation details, as well as the data collection procedures.

The purpose of the monitoring program is to identify physical changes to the conditions of the facilities which may lead to future instability and to allow Minto to mitigate these conditions prior to any occurrence of instability. The facilities have been separated into two sets of geotechnical thresholds and response criteria (Table 2-17). Soil cover performance and stability is addressed under Section 2.5 General Reclamation Measures of this plan.

Table 2-17: Physical Stability Categories

Category	Facility
1	• Dry Stack Tailings Storage Facility and Mill Valley Fill Extension (Stage 1 and 2) • Southwest Waste Dump • South Wall Buttress of the Main Pit Dump
2	• Main Waste Dump and Main Waste Dump Extension • Reclamation Overburden Dump • Ice-Rich Overburden Dump • Main Pit Dump (except the South Wall Buttress) • Area 118 Backfill Dump

Category 1 facilities are founded in areas of ice-rich periglacial foundations that have previously experienced deep seated foundation movement. The Mill Valley Fill Extension (MVFE) and South Wall Buttress (SWB) are designed to mitigate movements in the Dry Stack Tailings Storage Facility (DSTSF) and Main Pit South Wall areas, respectively, and the Main Pit Dump provides further stabilization to the Main Pit South Wall area. Additional monitoring inspection and response requirements for the DSTSF are detailed in the Operations, Maintenance, and Surveillance Manual (OMS) for the facility (Minto Explorations Ltd., 2014).

Category 2 facilities consist of all the remaining waste rock dumps. These waste dumps are located in areas with good foundation conditions that avoid areas underlain by ice-rich overburden.

2.4.2 Risk Narrative

A mass failure of one of the waste facilities has the potential to endanger the health and safety of site employees or visitors, or lead to an increase in contaminant loadings from the mine and subsequent adverse effects to aquatic resources in the receiving environment (lower Minto Creek).

2.4.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to Category 1 and Category 2 Facilities are provided in Table 2-18 and Table 2-19, respectively. The responses are dependent on the phase of closure as indicated in the tables. The monitoring results that are evaluated and utilized for this component of the AMP are a requirement of the Physical Monitoring Program of the EMSRP (Minto Explorations Ltd., 2016) and the Monitoring and Maintenance section (7.11) of the Minto RCP.

Table 2-18: Specific Indicators, Performance Thresholds and Responses for Category 1 Facilities

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses

• Mass movement indicated by monitoring of geotechnical instrumentation
• Visual observations of physical damage
• Visual observations of evidence that could suggest mass movement
• Occurrence of seismic events
	Specific Threshold 1
	• Observation of unusual occurrence including:
	  • tension cracks, settlement, or sloughing;
	  • a seismic event that exceeds the 1:475 return period event[2];
	  • abnormal seepage from any area of the slopes;
	  • increased turbidity from seepage;
	  • physical damage.
		All Stages	**Notification**
			• Mine/Project Manager
			• Geotechnical Engineer
			• Include in scheduled WUL reporting
			Review
			• Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data.
			Evaluation
			• Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability.
			Action
			• Inspect the area for any other signs of instability.
			• Follow any recommendations of the Geotechnical Engineer. At a minimum, the Engineer will consider the need for:
			  o An increase in the frequency of routine inspections and monitoring.
			  o Additional inspection, instrumentation, monitoring, or analyses
			• If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.

	Specific Threshold 2 • One survey hub or inclinometer reading indicating acceleration of movement greater than the long-term trend and outside range of instrumentation error	**All Stages**	**Notification** • Geotechnical Engineer **Review** • Review existing instrumentation data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results. **Action** • Retake reading. • If the reading was accurate, increase the survey hub or inclinometer frequency. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.
	Specific Threshold 3 • For DSTSF and MVFE piezometers 13-DSP-05a, 13-DSP-06, 15-DSP-07, and 15-DSP-08, an increase in pore water pressures such that Ru[3] exceeds 0.4[4]. • For MVFE2 piezometer 15-DSP-10, an increase in pore water pressures such that the water elevation is 1 m above the original ground surface. Or • Temperature greater than zero at a depth of 2 m below original ground *(all SWD ground temperature cables, and DSTSF ground temperature cables DST-10, DST-11, and DST-14 only)*	**Active** **Closure**	**Notification** • Mine Manager • Geotechnical Engineer • Include in scheduled WUL reporting **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability. **Action** • Inspect the area for any other signs of instability. • If pore water pressure threshold exceeded, increase frequency as directed by the Engineer until determined unnecessary. • Follow any additional recommendations of the Geotechnical Engineer. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.
		Post Closure I, **Post Closure II**	**Notification** • Project Manager • Geotechnical Engineer • Include in scheduled WUL reporting

[2] This size of a seismic event would be felt by most people on site. It would shake buildings, and rattle or break dishes, hanging objects, etc. Earthquake information may also be found online at: http://www.earthquakescanada.nrcan.gc.ca/index-eng.php

[3] Ru is the pore water pressure coefficient which is the ratio of pore water pressure to the overburden pressure. A pore water pressure ratio of 0.5 would be similar to the effect of a groundwater table at surface.

[4] Piezometer 13-DSP-5b exceeded this trigger on April 2015. A review of the data and stability analysis was completed and documented in the 2015 annual geotechnical inspection report (SRK 2015a) and found no stability issue and recommended continued monthly monitoring.

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
			Review • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability. **Action** • Inspect the area for any other signs of instability. • If pore water pressure threshold exceeded, increase frequency as directed by the Engineer until determined unnecessary. • Follow any additional recommendations of the Geotechnical Engineer. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.
	Specific Threshold 4 • For DSTSF and MVFE piezometers 13-DSP-05, 13-DSP-06, 15-DSP-07, and 15-DSP-08, an increase in pore water pressures such that Ru exceeds 0.6. • For MVFE2 piezometer 15-DSP-10, an increase in pore water pressures such that the water elevation is 3 m above the original ground surface.	**Active Closure**	**Notification** • Mine Manager • Geotechnical Engineer • Include in scheduled WUL reporting **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability. **Action** • Inspect the area for any other signs of instability. • Increase pore water pressure monitoring and data review frequency as directed by the Engineer until determined unnecessary. • Follow any additional recommendations of the Geotechnical Engineer. At a minimum, the Engineer will consider the need for: o An increase in the frequency of routine inspections and monitoring. o Additional inspection, instrumentation, monitoring, or analyses. o Modifications to the waste placement/construction practices. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.
		Post Closure I, Post Closure II	**Notification** • Project Manager • Geotechnical Engineer • Include in scheduled WUL reporting **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability. **Action** • Inspect the area for any other signs of instability. • Increase pore water pressure monitoring and data review frequency as directed by the Engineer until determined unnecessary. • Follow any additional recommendations of the Geotechnical Engineer. At a minimum, the Engineer will consider the need for: o An increase in the frequency of routine inspections and monitoring. o Additional inspection, instrumentation, monitoring, or analyses. o Modifications to the waste placement/construction practices. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.
	Specific Threshold 5 • Three consecutive survey hub or inclinometer readings indicating acceleration of movement greater than the long-term trend. Or • Three consecutive survey hub readings indicating a change in horizontal direction of movement greater than 15 degrees from the long term trend.	**Active Closure**	**Notification** • Mine Manager • Geotechnical Engineer • Include in scheduled WUL reporting **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability. **Action** • Inspect the area for any other signs of instability. • Complete a ground survey of the area of interest to monitor any future displacement. • Increase monitoring and data review frequency as directed by the Engineer until determined unnecessary. • Follow any additional recommendations of the Geotechnical Engineer. At a minimum, the Engineer will consider the need for: o An increase in the frequency of routine inspections and monitoring.

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
			o Additional inspection, instrumentation, monitoring, or analyses o Modifications to the waste placement/construction practices, including discontinuation of loading. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.
		Post Closure I, Post Closure II	**Notification** • Project Manager • Geotechnical Engineer • In Include in scheduled WUL reporting **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability. **Action** • Inspect the area for any other signs of instability. • Complete a ground survey of the area of interest to monitor any future displacement. • Increase monitoring and data review frequency as directed by the Engineer until determined unnecessary. • Follow any additional recommendations of the Geotechnical Engineer. At a minimum, the Engineer will consider the need for: o An increase in the frequency of routine inspections and monitoring. o Additional inspection, instrumentation, monitoring, or analyses o Modifications to the waste placement/construction practices, including discontinuation of loading. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately

Table 2-19: Specific Indicators, Performance Thresholds and Responses for Category 2 Facilities

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
• Mass movement indicated by monitoring of geotechnical instrumentation • Visual observations of physical damage • Visual observations of evidence that could suggest mass movement • Occurrence of seismic events	**Specific Threshold 1** • Observation of unusual occurrence including: • tension cracks, settlement, or sloughing; • a seismic event that exceeds the 1:475 return period event; • abnormal seepage from any area of the slopes; • increased turbidity from seepage; • physical damage.	**All Stages**	**Notification** • Mine/Project Manager • Geotechnical Engineer • Include in scheduled WUL reporting **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability. **Action** • Inspect the area for any other signs of instability. • Follow any recommendations of the Geotechnical Engineer. At a minimum, the Engineer will consider the need for: o An increase in the frequency of routine inspections and monitoring. o Additional inspection, instrumentation, monitoring, or analyses. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.
	Specific Threshold 2 • Survey hub cumulative displacements between 150 mm and 500 mm; Or • Increase in pore water pressures such that Ru exceeds 0.2.	**Active Closure**	**Notification** • Mine Manager • Geotechnical Engineer • Include in scheduled WUL reporting **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability. **Action**

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
			• Inspect the area for any signs of instability. • Immediately Increase monitoring and data review frequency until determined unnecessary: o If survey hub threshold exceeded, increase frequency as directed by the Engineer. o If pore water pressure threshold exceeded, increase frequency as directed by the Engineer. • Follow any additional recommendations of the Geotechnical Engineer. • If the results of the analysis indicates there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.
		Post Closure I, Post Closure II	**Notification** • Project Manager • Geotechnical Engineer • Include in scheduled WUL reporting **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability. **Action** • Inspect the area for any signs of instability. • Increase monitoring and data review frequency until determined unnecessary: o If survey hub threshold exceeded, increase frequency as directed by the Engineer. o If pore water pressure threshold exceeded, increase frequency as directed by the Engineer. • Follow any additional recommendations of the Geotechnical Engineer. • If the results of the analysis indicates there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.
	Specific Threshold 3 • Survey hub cumulative displacements greater than 500 mm. Or • Increase in pore water pressures such that Ru exceeds 0.4.	**Active Closure**	**Notification** • Mine Manager • Geotechnical Engineer • Include in scheduled WUL reporting **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability. **Action** • Inspect the area for any other signs of instability. • Complete a ground survey of the area of interest to allow for a stability assessment to be completed (if required by the Engineer), and to monitor any future displacement. • Increase monitoring and data review frequency until determined unnecessary: o If survey hub threshold exceeded, increase frequency as directed by the Engineer. o If pore water pressure threshold exceeded, increase frequency as directed by the Engineer. • Follow any additional recommendations of the Geotechnical Engineer. At a minimum, the Engineer will consider the need for: o An increase in the frequency of routine inspections and monitoring. o Additional inspection, instrumentation, monitoring, or analyses o Modifications to the waste placement/construction practices, including discontinuation of loading. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.
		Post Closure I, Post Closure II	**Notification** • Project Manager • Geotechnical Engineer • Include in annual report **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability. **Action** • Inspect the area for any other signs of instability. • Complete a ground survey of the area of interest to allow for a stability assessment to be completed (if required by the Engineer), and to monitor any future displacement. • Increase monitoring and data review frequency until determined unnecessary:

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
			o If survey hub threshold exceeded, increase frequency as directed by the Engineer. o If pore water pressure threshold exceeded, increase frequency as directed by the Engineer. • Follow any additional recommendations of the Geotechnical Engineer. At a minimum, the Engineer will consider the need for: o An increase in the frequency of routine inspections and monitoring. o Additional inspection, instrumentation, monitoring, or analyses o Modifications to the waste placement/construction practices, including discontinuation of loading. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.

2.5 General Reclamation Measures

2.5.1 Soil Cover Performance

2.5.1.1 Description

Soil covers are to be placed over the dry stacked tailings storage facility (DSTSF) and the waste rock storage facilities. Various waste rock storage facilities exist across the site including the Main Waste Dump (MWD), Main Waste Dump Expansion (MWDE), the Main Pit Dump (MPD), the Southwest Waste Dump (SWD), the High Grade Waste (HGW), the Area 118 Dump, and the Mill Valley Fill (MVF). An infiltration reduction soil cover is to be placed on these features either as part of the progressive reclamation program or during final closure activities. The cover on the HGW is to include a very low infiltration cover system incorporating a geosynthetic membrane. The primary cover functions are to:

- Minimize infiltration to the extent practical using locally available material;
- Ensure a stable landform that will promote establishment of natural vegetation endemic to the area; and
- Minimize ponding and surface erosion on the final landform.

The design criteria for the site is described in the Section 2.2 of the preliminary cover design report (SRK 2016). Design details specific to the DSTSF and each of the waste rock storage facilities are provided in Sections 5 thru 10 of the design report.

2.5.1.2 Risk Narrative

- Covers do not perform as designed resulting in unacceptable water quality conditions downstream of the site.
- Cover on a steeper slope becomes over saturated and slumps.

2.5.1.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to water management are provided below in Table 2-20. The monitoring results that are evaluated and utilized for this component of the AMP are a requirement of the Post-Closure Geotechnical Monitoring Program.

Table 2-20: Specific Indicators, Performance Thresholds and Responses for Cover Performance

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses

• Visual observations of physical damage
• Visual observation of ponding water
• Visual observation of surface erosion
	Specific Threshold 1
	• Observation of unusual occurrence including:
	  • settlement, or sloughing;
	  • abnormal seepage from any area of the slopes;
	  • increased turbidity from seepage;
	  • isolated pockets of vegetation stress;
	  • physical damage.
		All Stages	**Notification**
			• Mine/Project Manager
			• Geotechnical Engineer
			• Include in scheduled WUL reporting
			Review
			• Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data.
			Evaluation
			• Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability.
			Action
			• Inspect the area for any other signs of instability.
			• Repair the cover to adhere to design specifications.
			• Follow any recommendations of the Geotechnical Engineer. At a minimum, the Engineer will consider the need for:
			  o An increase in the frequency of routine inspections and monitoring.
			  o Additional inspection, instrumentation, monitoring, or analyses.
			• If the results of the analysis indicates there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.

	Specific Threshold 2 • Erosion of cover; • Large scale vegetation stress	**All Stages**	**Notification** • Mine Manager • Geotechnical Engineer • Include in scheduled WUL reporting **Review** • Review previous inspection reports, existing instrumentation including piezometer, temperature, inclinometer, and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability. **Action** • Inspect the area for any signs of instability. • Immediately increase monitoring and data review frequency as directed by the Engineer until determined unnecessary. • Follow recommendations of the Geotechnical Engineer to repair the cover. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.

2.5.2 Minto North Pit Water Level

2.5.2.1 Description

At closure, the Minto North Pit will be allowed to flood. Groundwater observations and preliminary water balance monitoring under most scenarios indicate the pit is not likely to fill to the point where it would spill. Monitoring of pit water levels have been included in the post-closure monitoring Program, in Section 7.11, Table 7-1. The water balance model will be reviewed and updated annually during the post closure, and should the water level in the pit be projected to overflow, an engineered spillway may be required to control outflow from the pit and mitigate erosion.

2.5.2.2 Risk Narrative

- Water levels rise higher than predicted by the site water balance model and overtop the pit resulting in erosion of surficial materials at the spill location and unacceptable water quality conditions downstream of the site.

2.5.2.3 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to water management are provided below in Table 2-21. The monitoring results that are evaluated and utilized for this component of the AMP are included in Section 7.11 of the RCP.

Table 2-21: Specific Indicators, Performance Thresholds and Responses for Minto North Pit Water Level

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
Updated water balance model indicates pit will overtop	**Specific Threshold 1** • Updated water balance model indicates Minto North Pit overtop within 10 years.	**All Stages**	**Notification** • Mine/Project Manager • Geotechnical Engineer • Include in scheduled WUL reporting **Review** • Review water balance inputs and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of the need for spillway design considerations. **Action** • Develop spillway design. • Follow any recommendations of the Geotechnical Engineer. At a minimum, the Engineer will consider the need for: o An increase in the frequency of monitoring. o Additional inspection, instrumentation, monitoring, or analyses.
	Specific Threshold 2 • Updated water balance model indicates Minto North Pit overtop within 3 years.	**All Stages**	**Notification** • Mine/Project Manager • Geotechnical Engineer • SFN and YG • Include in scheduled WUL reporting **Review** • Review water balance inputs and survey data. **Evaluation** • Geotechnical Engineer to compare recent monitoring results against older results for additional evidence of instability for subsequent design modifications. **Action** • Develop spillway design (if not already completed). • Construct the spillway within the three-year period (or implement plan to manage water to prevent uncontrolled release from the pit). • Follow any recommendations of the Geotechnical Engineer. At a minimum, the Engineer will consider the need for: o An increase in the frequency of monitoring. o Additional inspection, instrumentation, monitoring, or analyses.

2.5.3 Revegetation

The primary objective is the minimization of erosion (and corresponding sediment delivery) from sloped cover areas. Revegetation treatments at Minto are designed to achieve restoration and land-use objectives, while achieving the protection objective of maintaining surficial substrate stability and erosion control where necessary.

2.5.3.1 Risk Narrative

- Vegetation does not establish fast enough to prevent rill erosions on covers
- Deep rooted vegetation not established before a forest fire.

2.5.3.2 Specific Indicators, Performance Thresholds and Responses

Indicators, performance thresholds and responses specific to water management are provided below in Table 2-22. The monitoring results that are evaluated and utilized for this component of the AMP are a requirement of the Monitoring and Maintenance program of this RCP.

Table 2-22: Specific Indicators, Performance Thresholds and Responses for Vegetation Performance

Specific Indicators	Specific Performance Thresholds	Stage	Specific Responses
• Visual observations of failure of vegetation to establish • Visual observation of failure of native species to establish. • Visual observation of stressed vegetation.	**Specific Threshold 1** • Observation of isolated pockets of stressed vegetation.	**All Stages**	**Notification** • Mine/Project Manager • Include in scheduled WUL reporting **Review** • Review previous inspection reports. **Evaluation** • Compare recent monitoring results against older results for additional evidence of vegetation stress. • Evaluate cover integrity. **Action** • Follow any recommendations of a qualified professional to determine if a reapplication of seeds and fertilizer is required. • Repair covers as recommended by a Geotechnical Engineer
	Specific Threshold 2 • Large scale stressed vegetation. • Failure of vegetation to establish in erosion-control areas.	**All Stages**	**Notification** • Mine Manager • Geotechnical Engineer • Include in scheduled WUL reporting **Review** • Review previous inspection reports and precipitation data. **Evaluation** • Compare recent monitoring results against older results for additional evidence of vegetation stress. • Evaluate cover integrity **Action** • Inspect the area for any signs of a source of contamination. • Immediately increase monitoring and data review frequency as directed by a qualified professional until determined unnecessary. • Follow recommendations of a qualified professional to restore vegetation to the area. • Repair covers as recommended by a Geotechnical Engineer. • If the results of the analysis indicate there is a stability concern, the mine inspector and Selkirk First Nation will be notified immediately.

3 Reporting and Review

Reporting and review represent an essential part of the Adaptive Management Framework, as described in section 1.3.2, and reporting on adaptive management is included in the notification component of all specific responses described in section 2.

3.1 Monthly and Annual Reporting

Monthly reports are required to be submitted to the Yukon Water Board under Water Licence QZ14-031, and annual reports are required for submission under both the Water Licence and the Quartz Mining License. Both licenses require reporting on adaptive management.

Monthly reporting includes all activities carried out under the Adaptive Management Plan and will be continued throughout active closure but will reduce frequency to annually for PCI and PCII.

Annual reporting will include summaries of all activities carried out under the Closure Adaptive Management Plan, including monitoring results compared to thresholds and any actions taken.

3.2 Annual Review

The AMP may be modified when unexpected circumstances are encountered and the protocol is implemented or when additional understanding becomes available. An annual review of the AMP will take place prior to annual reporting, and Annual Reports will include a summary of proposed updates and revisions to the Adaptive Management Plan and include a revised Adaptive Management Plan, if warranted.

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Minto Mine Closure Cost Estimate – RCP Revision 2018-01

Prepared for

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Appendix A – Basis and Validation of Labour and Equipment Rates
Appendix B – Cost Estimate Worksheets

1 Introduction

1.1 General

This document presents the estimates for reclamation and closure liability based on the Minto Mine Reclamation Closure Plan, revision 2018-01. The estimates have been prepared in accordance with the costing guidance from YG EMR in the document Reclamation and Closure Planning for Quartz Mining Projects: Plan Requirements and Closure Costing Guidance (YG, 2013).

The report describes the basis of the estimate and the major costing assumptions. This document should be read in conjunction with the entire RCP, which provides details on the level of engineering for each cost component, as well as the cost estimate spreadsheet provided in Appendix B, which provides a complete costing details and assumptions.

1.2 Scope of Estimate

YG (2013) guidance document requires liability estimates for three separate conditions:

- Current status (i.e. Year 0);
- Peak Liability within the two-year period for this the RCP approval will apply; and,
- End-of-mine (EOM) Life.

For the Minto Mine, the peak liability is expected to occur in 2019 following mining of the Ridgetop North and Ridgetop South Open Pits. The Year 0 liability estimate assumes that mining at the Area 2 Stage 3 Pit is complete, and that mining of the Ridgetop North and South has yet to begin. The end-of-mine life closure liability occurs following completion of all open pit and underground mining, as well as the milling of all ore stockpiles by the end of 2021.

Closure implementation is estimated to take 3 years for each liability estimate with construction occurring between April and September. The site is estimated to be accessed by air during the spring, and by barge following spring break-up. Two years of interim care and maintenance is assumed to be required after mining ceases and prior to implementation of the final closure plan for the Year 0 and Peak liability estimates. These costs are not included in the EOM estimate as the closure plan would be finalized at the end-of-mining.

As noted in Section 8 of the RCP, the Post-Closure I and Post-Closure II periods have durations of 5 years and 10 years, respectively. During Post-Closure I, site access is maintained by air during the spring, and by barge following spring break-up. During Post-Closure II, site maintenance is assumed to be required every 5 years with the site accessed by barge and work completed in one month. Site access for monitoring events and geotechnical inspections when no barge is in place are assumed to be completed by helicopter. During Post-Closure II, the geotechnical inspection completed by helicopter will also be used to determine the maintenance requirements to be completed during the next summer.

Costs have also been allocated for on-going site maintenance past the Post-Closure II period into perpetuity. Routine site inspections and maintenance were assumed to be completed every 10 years, with access to the site established by barge. During these 10 year maintenance events, costs were allocated for passive treatment maintenance, site grading, as well as maintenance of the access road (including Big Creek Bridge).

2 Unit Rates

2.1 Equipment Rates

The equipment rates used in the Minto cost estimate are based on 2017-18 BC Blue Book equipment rates[1]. The BC Blue Book rates are determined by the BC Ministry of Transportation based on recommendations from the BC Road Builders and Heavy Construction Association, and is validated from its own research and calculations. The BC Blue book publishes 'all-found rates' with includes all ownership and operating costs that consist of the following:

- Ownership: financing costs and depreciation
- Maintenance: repair, lube, wear items, and tire costs
- Operator costs: wages including benefits and overhead
- Equipment overhead and profit.

The BC Blue Book also publishes 'all-found less operator' rates that exclude operator costs.

As many of the above items are site-specific and are included in other portions of the cost estimate (labour, overhead, profit, fuel), these costs are removed from the "all-found" rate to obtain the base ownership and maintenance (O&M) hourly rates to avoid double-accounting of costs. The BC Rates were selected for use as a basis in the cost estimate as the methodology to obtain the base O&M rate from the published 'all-found' hourly rates is known, and use of the O&M rates in the cost estimate allows for site-specific mark-ups to be applied in a clear and transparent way. The details of the calculation details and mark-ups applied are provided in Section 3.2.

Use of the O&M rates in the cost estimate is a robust methodology to develop the overall project costs as the cost of ownership and maintenance is not expected to vary significantly between project locations, and allows for the site-specific project mark-ups to be costed based on the expected execution of the actual project.

The BC Blue Book 'all-found less operator' unit rates were broken down to obtain the base ownership and maintenance costs for each piece of equipment as follows:

- 20% profit and overhead was removed. This percentage mark-up was obtained from communications with the BC Road Builders and Heavy Construction Association.

[1] B.C. Road Builders and Heavy Construction Association.

- Fuel was removed based on a $0.94/L and based on fuel consumption rates obtained from the Caterpillar Performance Handbook. The fuel rate to remove was obtained from the BC Ministry of Transportation website.

A site-specific fuel operating cost was then added back to the equipment rates based on a delivered diesel fuel cost of $0.96/L was assumed based on the average on-site cost. The equipment costs are further marked-up in the project indirect costs where contractor profit (10%), contract administration (5%), bonding (3%), insurance (1.5%) are applied, resulting in a further mark-up of 19.5%.

Appendix A provides a comparison of the BC Blue Book rates to rates listed in the Yukon Government Equipment Rental Rates Guidebook. This comparison and review of equipment rates was completed in 2017 as a result of review comments of the 2016 Minto Mine Closure Cost Estimate (RCP 2016-01) received from the Yukon Government Department of Energy, Mines and Resources. The results of the comparison recommended that the equipment rate basis presented in this section continue to be used as there is a valid basis to the derived 'ownership and maintenance' equipment costs, and the site-specific build-up of the fuel and indirect costs results in an overall project cost that is representative to the execution of the Reclamation and Closure Plan.

2.2 Labour Rates

Labour rates for a third-party contractor were built up from base hourly rates presented in the Yukon Government Fair Wage Schedule effective April 1, 2017. The resulting rates were compared and validated with actual on-site base wages, and wages published by InfoMine (2017) for another mine site in the Yukon Territory. Details of the comparison are provided in Appendix A.

Workers were assumed to work a '3 weeks in, 2 weeks out' schedule, working 12 hour days while on site. The base wages were adjusted to account for: travel, overtime, vacation, statutory holidays, and other benefits as follows:

- **Travel Time:** Workers were compensated for travel time at the start and end of each rotation. Staff were assumed to originate either from Pelly Crossing or Whitehorse, and were compensated either 2.5 or 4 hours, respectively, for each trip between home origin and the site.
- **Overtime:** Each hour worked in addition to a standard 40 hour work week was compensated at a rate 1.5 times the base hourly wage.
- **Vacation**: Workers were compensated a pro-rated amount of 80 hours of pay over each year worked.
- **Statutory Holidays**: Workers were compensated a pro-rated amount of 88 hours of pay over each year worked.

- **Employment Insurance (EI)**: EI was added at a rate of 1.6% of base earnings up to the maximum insurable earning of $51,300 as per Canadian Revenue Agency guidance.[2]
- **Canada Pension Plan (CPP):** CPP contributions were added at a rate of 4.95% of base earnings up to the CPP salary limit of $55,300.[3]
- **Worker's compensation**: Added for all workers at a rate prescribed by the Yukon Workers' Compensation Health and Safety Board (YWC) that varied depending on the worker's group assessment rate as determined by YWC. Rates varied from 1.42% for camp support staff to 4.79% for equipment operators.[4]
- A small tool allowance was added at $250 per year for trades workers based on InfoMine (2017) for a British Columbian mine site.

The labour rates do not include costs of camp accommodation or flights that are included elsewhere in the cost estimate indirect costs. In addition, labour rates do not include MSP, as the Yukon Territory has no employer or employee premiums or payroll taxes for health care.

Including all benefits paid by the third-party contractor, the 'loaded' labour rates are generally 45 to 60% higher than the base hourly wage depending on the labour category.

2.3 Material Rates

Primary material rates used in the estimate consisted of: revegetation materials, fuel, and geosynthetic materials. The revegetation, fuel, and geotextile costs were provided by supplier quotes. The remaining material costs were obtained using RSMeans Online – a cost database service that provides up-to-date construction cost data. Material costs used were factored by RSMeans to be based in Whitehorse, and were further adjusted by 15% to account for shipping to Minto.

3 Basis of Estimate – Closure

3.1 Direct Costs

Direct costs were estimated based on the construction fleet and productivities assigned to each task. Determination of the construction fleet and productivities were obtained based on first principals, from experience on similar projects, or from the RS Means Online cost database. Earthmoving unit rates were based on calculated productivities that followed standard methods, as used by earthwork contractors. The calculations make use of equipment specifications obtained from manufacturer's data, in this case the Caterpillar Handbook.

Table 1 provides a summary of the major direct cost quantity sources and assumptions. Further details are provided in the cost estimate worksheets provided in Appendix B.

[2] http://www.cra-arc.gc.ca/tx/bsnss/tpcs/pyrll/clcltng/ei/cnt-chrt-pf-eng.html
[3] http://www.cra-arc.gc.ca/tx/bsnss/tpcs/pyrll/clcltng/cpp-rpc/cnt-chrt-pf-eng.html
[4] http://www.wcb.yk.ca/QuestionResults/Assessments/Rates/Q0269.aspx

Table 1: Direct Cost Inputs and Costing Assumptions

Area	Inputs and Assumptions
Waste Dumps	• Waste rock dump and ore stockpile dimensions were obtained from AutoCAD drawings of the September 2017 site topography with dimensions for waste dump facilities currently under construction estimated based on design drawings. • The dump slopes were assumed to be regraded to be flatter than 3H:1V and flat areas were assumed to be regraded to form tertiary drainage catchments with a minimum 2% grade. • The SWD high grade waste stockpile was assumed to be relocated to the Main Pit and placed below the expected long-term water elevation of 786 m. The remaining stockpile volume was provided by Minto on December 31, 2017. • The waste rock covers were assumed to be entirely sourced from the Reclamation Overburden Dump.
Overburden Dumps	• Dump dimensions were obtained from AutoCAD drawings of the September 2017 site topography. • The dump slopes were assumed to be regraded to be flatter than 3H:1V and flat areas were assumed to be regraded to form tertiary drainage catchments with a minimum 2% grade.
Ore Stockpiles	• The remaining stockpile volumes near the mill and the Main Waste Dump were provided by Minto on December 31, 2017. They are assumed to be relocated into the Main Pit, along with the upper two meters of the ore pads. • Stockpile pad dimensions were obtained from AutoCAD drawings of the September 2017 site topography.
Open Pits	• Earthen safety berms are assumed to be placed around the pit high walls, with large boulders placed near pit access points and warning signs placed at key locations around the perimeter. • Perimeter lengths were obtained from the latest site topography for completed pits, or design drawings for pits currently under construction, or yet to be mined.
Underground Openings	• A backfill plug is assumed to placed at all portals with backfill material sourced from near-by waste rock. The areas near the portal is assumed to be backfilled and recontoured. • All vent raises are assumed to be sealed with a reinforced concrete cap with a vent raise pipe, and covered with fill.
External Tailings Facilities	• Dimensions of the DSTSF were obtained from AutoCAD drawings of the September 2017 site topography.
Roads	• Road dimensions were obtained from the 2017 site topography and aerial photos. Roads no longer required following closure are assumed to be scarified and revegetated. • The main site access road is assumed to be maintained for long-term site access.

Area	Inputs and Assumptions
Demolition	• Building dimensions were obtained from site drawings for all major structures, and were estimated based on site photos for minor structures where drawings were not available. • All equipment was assumed to be dismantled and removed from site. Equipment lists were provided by Minto in 'Issued-for-Construction' Drawings, or based on the Hatch 2006 feasibility study equipment list. • Demolition debris was assumed to be disposed in the on-site landfill located near the airstrip, with all hazardous material, recyclables, and re-useable equipment removed an appropriate off-site facility.
Site infrastructure	• Pipeline and power line lengths and details were provided by Minto in AutoCAD format. • The pipelines were assumed to be disposed on-site, with the power lines and power poles disposed off-site once no longer needed.
Water Detention Structures	• The water storage dam volumes were obtained from dam as-built drawings. The dam material is assumed to be partially utilized as construction material for the wetland system upstream of the dam.
Yards	• Yard areas were obtained from AutoCAD drawings and aerial photography. The yard boundaries are provided in Figure 9-1.
Waste Disposal	• Hydrocarbon contaminated soils were assumed to be hauled to the on-site landfill located at the airstrip, with the landfarm operated for a period of three years. • Areas of hydrocarbon contaminated soils were assumed to be present near fuelling storage and mechanic working areas, with an average contamination depth of 1.0 m.
Surface water conveyance	• Excavation volumes, channel lengths, alignments, and other quantities for the primary catchment channels were obtained from AutoCAD drawings. • The secondary catchment quantities were calculated based on the typical section provided in Section 7 of the RCP, with an allowance of 40 m of secondary channel lengths per hectare of dump.
Water Treatment	• Fill used to construct the wetland cells was assumed to be sourced from the Water Storage Pond Dam. • The wetland cells are assumed to be lined with HDPE with organics assumed to be produced using a power mulcher form vegetation cleared from the water storage pond footprint. • The passive treatment system is assumed to be constructed in Year 3 of closure with the first year of operation in Year 1 of post-closure. • The active treatment system is assumed to be decommissioned and removed from site at the end of Post-Closure 2.

3.2 Indirect Costs

Indirect costs during the closure implementation include costs for mobilization-demobilization, site transport costs, site and road maintenance, construction support, project management, and quality assurance. A summary of the major indirect assumptions is provided in Table 2. Further

details regarding assumptions and cost sources are provided in the cost estimate worksheets in Appendix B.

Table 2: Indirect Cost Inputs and Costing Assumptions

Area	Inputs and Assumptions
Mobilization	• The assumed closure equipment fleet is listed in Worksheet 24 in Appendix B. Each piece of equipment is likely not required throughout the entire closure period, however, for the purposes of the estimate, the entire fleet was assumed to be mobilized in Year 1 of closure, with most equipment demobilized from site at the end of Year 3. • A small site maintenance fleet was assumed to remain on site to allow for earthwork repairs and passive treatment maintenance during the post-closure period. The storage location of the equipment has yet to be finalized, but is expected to consist of an existing site warehouse. • All equipment was assumed to be mobilized from Whitehorse.
Site Transport	• Site transport costs include barge and bus operations, as well as air transport and airstrip operations during periods where the barge is not in service.
Road and Site Maintenance	• A water truck and grader were both assumed to be required 150 hours per month. • An allowance of 7 km of silt fencing and 1 ha of erosion control matting was assumed for erosion protection prior to vegetation begin established.
Construction Support	• Field support staff, vehicles, and support equipment were estimated based on past project experience on projects of similar size. • The existing camp and office facilities are assumed to be used by the closure contractor. • Camp cost man-days were estimate based on a sum of the calculated man-hours required for closure implementation. • Camp costs including power and heat were provided by Minto.
QA, Project Management	• Project management and QA during implementation includes costs for staffing to provide on-site management of the contractor to ensure the project is implemented as per the closure plan.
Indirect Percentage Add-on Costs	• Additional indirect add-ons included the following: o Contingency = 12% o Worker's Compensation = 0% (included in labour rates) o Insurance = 2% of labour costs (equipment insurance included in equipment rates). o Bonding = 3% of project implementation and care and maintenance costs. o Contractor profit = 10% o Contract administration = 5%

3.3 Contingency

As noted in Table 2, contingency is added to all costs at a rate of 12%. In Worksheet 1 (Cost Summary), the implementation cost contingency is applied to the direct costs (A01 through A14), interim monitoring and care and maintenance, and Indirect Costs (B1 through B5). The closure cost contingency is applied to the direct cost items 1 to 7 in Worksheet 1. Contingency is also

distributed in the cost schedules and subjected to inflation and net present value calculations as detailed in Section 6.

The contingency value was selected based on a review of the site uncertainty and available published reclamation guidance. The contingency value was lowered from 15% to 12% in the 2016 version of the RCP because of advancement in the engineering design (i.e. landform design, wetland treatment, and conveyance channels) and reduction in the performance uncertainty related to the DSTSF and Mill Valley Fill Extension Stage 2. RECLAIM guidance recommends a contingency of 10% for definitive or construction level drawings, and 15% for preliminary or budget level drawings (GNWT 2017). The level of engineering for the Minto closure is believed to be between these two values. 12% was selected because the engineering design has already adopted generally conservative assumptions where uncertainty remains.

4 Basis of Estimate – Post Closure and C&M Costs

The following section describes the costs allocated for care and maintenance of the site for all phases of closure. Further details of the costs for each closure phase are provided in Worksheets 3, 8 and 13 for the Year 0, Peak, and EOM liability estimates, respectively, in Appendix B.

4.1 Water Treatment

Active treatment operational costs are based on on-site costs for 2015. Based on a review of water treatment costs from 2013 through 2016, 2015 was found to best represent typical conditions in terms of days operating per year and operating costs. The 2015 costs were updated to 2018 dollars based on an inflation factor of 1.03 obtained from the Bank of Canada inflation calculator. The active treatment system is assumed to be operated through Active Closure and Post-Closure I. Costs for capital replacement and maintenance of the active treatment system are included through to the end of the Post-Closure II period, after which the system is assumed to be dismantled and removed from site.

The passive treatment wetlands are assumed to start operations in Year 1 of post-closure. Allocated costs for operation and maintenance of the system includes costs for staffing and oversight of the system, an allowance for replacement of the organic media, as well as for carbon source injections.

4.2 Reclamation and Site Maintenance

Reclamation and site maintenance costs include allowances for equipment maintenance, road maintenance, repair of earthworks including site grading, cover and revegetation repair.

4.3 Site Management

Costs are included for the management of the site and care and maintenance activities, as well as camp operation.

4.4 Transport Costs

Transport costs include staffing transport between site and Whitehorse, as well as barge operations and mobilization-demobilization. Barge costs are based on Minto's operational site costs and assumed to be operated 20 hrs per week.

4.5 Post-Closure Indirect Costs

Post-closure indirect costs include the same percentage add-on costs as used during closure. These costs include: contingency (12%), insurance, bonding, contractor profit, and contract administration.

5 Closure Planning, Engineering, Permitting, and Monitoring

Planning, permitting and engineering cost details are provided in Worksheets 6, 11, and 16 for the Year 0, Peak, and EOM liability estimates, respectably.

Planning and permitting costs are included for the Year 0 and Peak liability estimates, and are assumed to be included during operations in the End-of-Mine Scenario. Planning and permitting costs include: reclamation research and planning, and allowance for additional technical studies, monitoring and management plans, and permitting staffing and meetings.

Engineering, Design, and Construction plans are calculated as 5% of the direct implementation costs. Costs for engineering oversight during construction is included in the indirect costs during implementation (i.e. on Worksheets 4, 9, and 14).

Monitoring of the site was costed according to the monitoring schedule presented in Section 7 of the RCP. Monitoring costing details are provided in Worksheet 25 in Appendix B.

6 Inflation and Net Present Value Analysis

Inflation and net present value (NPV) calculations are presented in Worksheets 2, 7, and 12 of Appendix B for the Year 0, Peak, and EOM liability estimates respectively. The tables include the cost schedule and undiscounted cash flow starting in 2018 through post-closure year 105 (100 years after Post-Closure 1). As required by YG (2013a), inflation has been applied to the period of active implementation, while NPV discount rates were applied to post-closure monitoring and maintenance costs. The NPV calculations include annual cash flows to perpetuity (modeled as 105 years after decommissioning and reclamation).

Inflation has been applied at a rate of 1.6%, which corresponds to the average annual rate of inflation for the past ten years.

As per Yukon Mine Site Reclamation and Closure Policy Financial Guidelines (YG 2013b), the discount rate used in NPV calculation are to be the most recent Government of Canada benchmark bond yields as published by the Bank of Canada, with the bond term selected to be

the longest term published that does not exceed the expected duration of the post-closure reclamation, monitoring and maintenance program. For the Minto site, this corresponds to a bond term of ten years, which as of February 2018, this corresponds to a bond yield of 2.35%. Given the historically low yield, a variable discount rate has been selected for the NPV calculations. The NPV calculations apply a discount rate of 2.35% for the first two years, 2.5% for the next 3 years, and 2.8% thereafter. The 2.8% corresponds to the average yield since 2005.

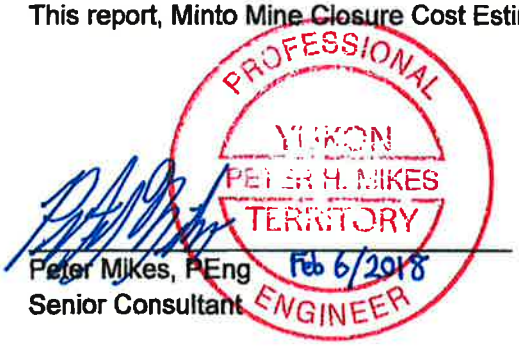
7 Cost Summary

Table 3 summarizes the reclamation and closure liability for the Year 0, Peak, and End-of-Mine scenarios. The costs are presented in 2018 Canadian Dollars.

Table 3: Reclamation and Closure Liability Cost Summary

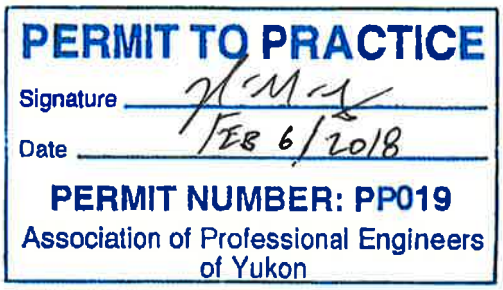
Description of Cost	Year 0	Peak	EOM
Closure Implementation			
Direct Costs	$17,044,478	$19,431,701	$16,987,765
Care & Maintenance Costs to the end of implementation	$10,152,363	$10,271,725	$3,910,414
Indirect Costs	$18,869,092	$20,172,562	$17,445,555
Cost Inflation	*$2,352,313*	*$3,724,678*	*$3,997,740*
Sub-total - Implementation Costs	$48,418,245	$53,600,666	$42,341,475
Post-Closure NPV	$13,777,523	$13,096,047	$12,621,082
TOTAL FINANCIAL SECURITY	**$62,195,768**	**$66,696,713**	**$62,701,817**

This report, Minto Mine Closure Cost Estimate – RCP Revision 2018-01, was prepared by



Peter Mikes, PEng
Senior Consultant

Feb 6/2018



and reviewed by

Dylan MacGregor, PGeo
Principal Consultant

All data used as source material plus the text, tables, figures, and attachments of this document have been reviewed and prepared in accordance with generally accepted professional engineering and environmental practices.

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The opinions expressed in this report have been based on the information available to SRK at the time of preparation. SRK has exercised all due care in reviewing information supplied by others for use on this project. Whilst SRK has compared key supplied data with expected values, the accuracy of the results and conclusions from the review are entirely reliant on the accuracy and completeness of the supplied data. SRK does not accept responsibility for any errors or omissions in the supplied information, except to the extent that SRK was hired to verify the data.

8 References

B.C. Road Builders and Heavy Construction Association, 2017. 2017-18 Equipment Rental Rate Guide. Published July.

[GNWT] Government of Northwest Territories, 2017. Reclaim 7.0 User Manual, Mining Version., Revised November 2017.

InfoMine Inc. (2017). CostMine Mining Cost Service, October.

[YG] Government of Yukon, Energy Mines and Resources, 2013a. Reclamation and Closure Planning for Quartz Mining Projects: Plan Requirements and Closure Costing Guidance. August.

[YG] Government of Yukon, Energy Mines and Resources, 2013b. Yukon Mine Site and Reclamation Closure Policy, Financial and Technical Guidelines September.

[YG] Government of Yukon, Department of Community Services, 2017. Fair Wage Schedule, effective April 1, 2017. Accessed at http://www.community.gov.yk.ca/fairwage.html on January 27, 2018.

Appendix A – Basis and Validation of Labour and Equipment Rates

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Memo

To:	Ryan Herbert	**Client:**	Minto Explorations Ltd.
From:	Peter Mikes, PEng.	**Project No:**	1CM002.045
Cc:	Dylan MacGregor, SRK	**Date:**	January 27, 2018
Subject:	Basis and Validation of Labour and Equipment Unit Rates used in the Minto Mine Closure Cost Model		

1 Introduction

Minto Mine's Reclamation and Closure Plan (RCP) Revision 2016-01 and closure cost estimate was submitted to the Yukon Water Board (YWB) in August 2016. Following a review of the RCP Revision 2016-01 closure cost estimate and memorandum by SteveJan Consultants Inc. (SJCI) on behalf of the Yukon Government Department of Energy, Mines and Resources, additional details on the basis and calculation of the labour and equipment rates were requested.

Additional basis for the rates and calculation details have been provided within the main body of this report. This memorandum provides a validation of the labour and equipment rates used in the 2016 cost estimate that have continued to be basis for the estimate for Revision 2018-01 of the RCP.

2 Labour Rates

2.1 Basis and Validation

Labour rates for a third-party contractor were built up from base hourly rates presented in the Yukon Government Fair Wage Schedule effective April 1, 2016. The Fair Wage Schedule was compared and validated with two sources: actual contractor base wages paid at site[1], and from a mine wage benefit survey published by InfoMine (2016)[2] for a "Tungsten mine located in the Yukon Territory"[3]. All three sources resulted in similar base hourly wages. As an example for the two published sources: InfoMine (2016) lists hourly base wages of $19.20 and $22.15 for entry-level and semi-skilled labourers, and $28.20 for a heavy equipment operators. The Yukon Fair Wage Schedule lists slightly higher hourly base wages of $23.19 for labourers, and $28.81 for heavy equipment operators. The Yukon Fair Wage Schedule was therefore deemed to be appropriate for use in the closure cost estimate.

[1] Personal Communication with Minto Mine Manager on May 23, 2017.
[2] InfoMine (2016), CostMine Mining Cost Service, Section LA Labour – Canada 2016.
[3] As there are no Tungsten Mines in YT, it is likely that the InfoMine reference is the Cantung Mine in NWT.

3 Equipment Rates

3.1 Basis

The equipment rates used in the Minto cost estimate are based on BC Blue Book equipment rates[4]. The BC Blue Book rates are determined by the BC Ministry of Transportation based on recommendations from the BC Road Builders and Heavy Construction Association, and is validated from its own research and calculations. The BC Blue book publishes 'all-found rates' with includes all ownership and operating costs that consists of the following:

- Ownership: financing costs and depreciation
- Maintenance: repair, lube, wear items, and tire costs
- Operator costs: wages including benefits and overhead
- Equipment overhead and profit.

The BC Blue Book also publishes 'all-found less operator' rates that exclude operator costs.

As many of the above items are site-specific and are included in other portions of the cost estimate (labour, overhead, profit, fuel), these costs are removed from the "all-found" rate to obtain the base ownership and maintenance (O&M) hourly rates to avoid double-accounting of costs. The BC Rates were selected for use as a basis in the cost estimate as the methodology to obtain the base O&M rate from the published 'all-found' hourly rates is known, and use of the O&M rates in the cost estimate allows for site-specific mark-ups to be applied in a clear and transparent way. The details of the calculation details and mark-ups applied are provided in Section 3.2.

Use of the O&M rates in the cost estimate is a more robust methodology to develop the overall project costs as the cost of ownership and maintenance would not be expected to vary significantly between project locations, and it allows for the site-specific project mark-ups to be costed based on the expected execution of the project.

3.2 Comparison with Yukon Guidebook Rates

Table 1 provides a comparison of the 'wet' (all-found) rates from the Yukon Guidebook compared to BC Blue Book rates for common equipment used in the closure cost estimate. For each type of equipment, Table 1 includes the number of unique pieces of equipment available in the Yukon Guidebook

Rates in Yukon Guidebook are highly variable and the basis for the rates are unknown. The Yukon Guidebook is prepared by the YG Department of Highways and Public Works and is understood to be compiled through a voluntary survey of contractors throughout the Territory. The Yukon Guidebook is understood to generally be used for smaller highways projects that are

[4] B.C. Road Builders and Heavy Construction Association.

not procured through a competitive public bidding process. In addition, there is no validation of the rates to determine the 'fair rate value' as is completed for the BC Blue Book rates.

Due to the variability in the costs, the Yukon Guidebook is not considered to be a valid source to determine the true O&M costs as the basis for the profit, overhead, and operator costs are unknown. The rates provided through the survey could be substantially different than what would be provided through a competitive bid process, or through any sort of negotiation. In addition, there are no listings for many of the larger pieces of equipment.

Table 1: Comparison between BC Blue Book and YG Equipment Rental Rates Guidebook Rates

	BC Blue Book 2015-16	YG Equipment Rental Rates 2016				Percentage difference between average YG and BC Rate
Equipment	All Found Rate ($/hr)	# of Listings	Minimum	Average	Maximum	
Dozer - CAT D8	$261.00	13	$200.00	$325.00	$455.00	25%
Dozer - CAT D10	$382.00	0				
Excavator CAT 330	$200.00	5	$280.00	$295.00	$330.00	48%
Excavator CAT 385	$355.85	0				
Grader - 16M	$173.00	4	$165.00	$206.00	$310.00	19%
Truck - CAT 735	$180.00	1	$250.00	$250.00		39%
Truck - CAT 777	$331.00	0				
Loader - CAT 950	$151.00	6	$160.00	$180.00	$215.00	19%
					Average difference:	**30%**

The average rate listed in the Yukon Guidebook is approximately 30% higher than rates listed in the BC Bluebook and the average of the minimum rates listed in the Yukon Guidebook is approximately 10% higher than the BC Bluebook. However, this comparison does not consider the additional indirect costs included in the Minto cost estimate that would typically be factored into equipment rates on a typical large earthworks project. Examples of these indirect costs include:

- Road maintenance that consists of a grader and a water truck
- Field support staff including a foreman, mechanic, surveyor, medic/H&S supervisor
- Field support vehicles, including pick-up trucks, fuel truck, mechanic service vehicle ETV, and a passenger bus for transporting workers to and from their work locations at the start and end of each shift
- Field support equipment including: lighting towers, management plans, office overhead (supplies, communications, office administration, office power and heat).
- Insurance (1.5%)

- Bonding (3%)
- Contractor Profit (10%)
- Contract Administration (5%)

If these indirect costs were to be factored into the equipment rates, the resulting equipment rates used in the Minto cost estimate would be higher than those listed in the Yukon Guidebook.

On completion of this review and comparison to the Yukon Guidebook equipment rates, we recommend that Minto maintain the approach to costing of equipment rates as there is a valid basis to the derived 'ownership and maintenance' equipment costs, and the site-specific build-up of the fuel and indirect costs results in an overall project cost that is representative to the execution of the Reclamation and Closure Plan.

Appendix B – Cost Estimate Worksheets

Table of Contents

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

Table of Contents

Worksheet #	Title	Worksheet Name
1	Cost Summary	01-Summary
2	Year 0 Estimate - Cashflow Calculations	02-Yr0NPV
3	Year 0 Estimate - Annual C&M costs	03-Yr0CCM
4	Year 0 Estimate - Implementation Costs	04-Yr0Active
5	Year 0 Estimate - Schedule Details	05-Yr0Sched
6	Year 0 Estimate - Planning and Permitting	06-Yr0Planning
7	Peak Liability Estimate - Cashflow Calculations	07-PeakNPV
8	Peak Liability Estimate - Annual C&M costs	08-PeakCCM
9	Peak Liability Estimate - Implementation Costs	09-PeakActive
10	Peak Liability Estimate - Schedule Details	10-PeakSched
11	Peak Liability Estimate - Planning and Permitting	11-PeakPlanning
12	End-of-Mine Liability Estimate - Cashflow Calculations	12-EOMNPV
13	End-of-Mine Liability Estimate - Annual C&M costs	08-EOMCCM
14	End-of-Mine Liability Estimate - Implementation Costs	14-EOMActive
15	End-of-Mine Liability Estimate - Schedule Details	15-EOMSched
16	End-of-Mine Liability Estimate - Planning and Permitting	16-EOMPlanning
17	Labour Rate Calculations	17-LaborRate
18	Equipment Rate Calculations	18-EquipRate
19	Material Rates and Indirect Cost Inputs	19-MatRate
20	Task Unit Rate Calculations	20-TASK UNIT RATE
21	Relocation Unit Rates	21-UNIT RELOCATION
22	Area Calculations for Covers and Revegetation	22-Areas
23	Demolition	23-Demolition
24	Mobilization-Demobilization Calculations	24-MobDemob
25	Monitoring Costs	25-Monitoring
26	Ore Stockpiles and Misc. Volume Calculations	26-Ore-MiscVol
27	Open Pit Calculations	27-Pits
28	Resloping Quantity Calculations	28-Regrade
29	Revegetation Perscriptions	29-Reveg
30	Road Calculations	30-Roads
31	Surface Infrastructure Calculations	31-SurfInfra
32	Underground	32-UG
33	Waste Disposal	33-Waste Disposal
34	WaterStructures	34-Water
35	Water Treatment	35-Water Treatment

Worksheet 1 - Cost Summary

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

COST SUMMARY - YEAR 0 (Dec. 31, 2017)

WBS	Description of Cost	Proposed Cost
Closure Implementation		
Direct Costs		
A01	Waste Dump	$6,847,702
A02	Overburden Dumps	$370,112
A03	Ore Stockpiles	$1,419,486
A04	Open Pits	$177,927
A05	Underground Openings	$81,637
A06	External Tailings Facilities	$202,656
A07	Roads	$136,050
A08	Demolition	$1,976,253
A09	Surface Infrastructure	$169,527
A10	Water Detention Structures	$786,952
A11	Yards/Laydown Areas	$758,907
A12	Waste Disposal	$452,923
A13	Surface Water Conveyance	$3,181,261
A14	Water Treatment	$483,085
Subtotal - Direct Implementation		**$17,044,478**
Monitoring and C&M		
P	Planning, Permitting, Monitoring	$2,325,691
CM	Care and Maintenance	$7,826,673
Subtotal - Monitoring and C&M		**$10,152,363**
Indirect Costs		
B1	Mobilization/Demobilization	$324,077
B2	Transport Costs	$405,599
B3	Site/Road Maintenance	$757,001
B4	Construction Support	$5,054,174
B5	QA and Project Management	$1,668,691
-	Other Indirects	$6,428,205
-	Contingency (12%)	$4,231,344
Subtotal - Indirect Costs		**$18,869,092**
Cost Inflation		*$2,352,313*
SUB-TOTAL IMPLEMENTATION COSTS		**$48,418,245**
Post-Closure Costs (Undiscounted)		
Direct Costs		
1	Monitoring	$1,761,326
2	Water Treatment - Active Treatment	$4,446,180
3	Passive Treatment	$560,239
4	Reclamation Maintenance	$1,834,053
5	Site Maintenance	$1,670,177
6	Site Management	$4,798,081
7	Transportation Costs	$994,297
Subtotal Direct Costs		**$16,064,353**
Indirect Costs		
1	Indirects	$2,934,273
2	Contingency (12%)	$1,945,144
Subtotal - Indirects		**$4,879,417**
SUB-TOTAL - POST-CLOSURE COSTS		*$20,943,770*
POST-CLOSURE NPV		**$13,777,523**
TOTAL FINANCIAL SECURITY		**$62,195,768**

COST SUMMARY - Peak Liability (2019)

WBS	Description of Cost	Proposed Cost
Closure Implementation		
Direct Costs		
A01	Waste Dump	$6,893,723
A02	Overburden Dumps	$395,276
A03	Ore Stockpiles	$3,331,999
A04	Open Pits	$620,582
A05	Underground Openings	$81,637
A06	External Tailings Facilities	$57,291
A07	Roads	$136,050
A08	Demolition	$1,976,253
A09	Surface Infrastructure	$169,527
A10	Water Detention Structures	$786,952
A11	Yards/Laydown Areas	$758,907
A12	Waste Disposal	$452,923
A13	Surface Water Conveyance	$3,287,496
A14	Water Treatment	$483,085
Subtotal - Direct Implementation		**$19,431,701**
Monitoring and C&M		
P	Planning, Permitting, Monitoring	$2,445,052
CM	Care and Maintenance	$7,826,673
Subtotal - Monitoring and C&M		**$10,271,725**
Indirect Costs		
B1	Mobilization/Demobilization	$324,077
B2	Transport Costs	$419,142
B3	Site/Road Maintenance	$791,662
B4	Construction Support	$5,344,480
B5	QA and Project Management	$1,752,126
-	Other Indirects	$6,958,307
-	Contingency (12%)	$4,582,768
Subtotal - Indirect Costs		**$20,172,562**
Cost Inflation		*$3,724,678*
SUB-TOTAL IMPLEMENTATION COSTS		**$53,600,666**
Post-Closure Costs (Undiscounted)		
Direct Costs		
1	Monitoring	$1,761,326
2	Water Treatment - Active Treatment	$4,446,180
3	Passive Treatment	$560,239
4	Reclamation Maintenance	$1,834,053
5	Site Maintenance	$1,670,177
6	Site Management	$4,798,081
7	Transportation Costs	$994,297
Subtotal Direct Costs		**$16,064,353**
Indirect Costs		
1	Indirects	$2,931,730
2	Contingency (12%)	$1,945,144
Subtotal - Indirects		**$4,876,874**
SUB-TOTAL - POST-CLOSURE COSTS		*$20,941,227*
POST-CLOSURE NPV		**$13,096,047**
TOTAL FINANCIAL SECURITY		**$66,696,713**

COST SUMMARY - End-of-Mine Liability (Closure Yr 1 202

WBS	Description of Cost	Proposed Cost
Closure Implementation		
Direct Costs		
A01	Waste Dump	$6,987,586
A02	Overburden Dumps	$395,276
A03	Ore Stockpiles	$794,200
A04	Open Pits	$620,582
A05	Underground Openings	$81,637
A06	External Tailings Facilities	$57,291
A07	Roads	$136,050
A08	Demolition	$1,976,253
A09	Surface Infrastructure	$169,527
A10	Water Detention Structures	$786,952
A11	Yards/Laydown Areas	$758,907
A12	Waste Disposal	$452,923
A13	Surface Water Conveyance	$3,287,496
A14	Water Treatment	$483,085
Subtotal - Direct Implementation		**$16,987,765**
Monitoring and C&M		
P	Planning, Permitting, Monitoring	$769,108
CM	Care and Maintenance	$3,141,306
Subtotal - Monitoring and C&M		**$3,910,414**
Indirect Costs		
B1	Mobilization/Demobilization	$324,077
B2	Transport Costs	$419,142
B3	Site/Road Maintenance	$791,662
B4	Construction Support	$5,263,751
B5	QA and Project Management	$1,752,126
-	Other Indirects	$5,378,346
-	Contingency (12%)	$3,516,451
Subtotal - Indirect Costs		**$17,445,555**
Cost Inflation		*$3,997,740*
SUB-TOTAL IMPLEMENTATION COSTS		**$42,341,475**
Post-Closure Costs (Undiscounted)		
Direct Costs		
1	Monitoring	$1,761,326
2	Water Treatment - Active Treatment	$4,446,180
3	Passive Treatment	$560,239
4	Reclamation Maintenance	$1,386,205
5	Site Maintenance	$1,670,177
6	Site Management	$4,798,081
7	Transportation Costs	$994,297
Subtotal Direct Costs		**$15,616,504**
Indirect Costs		
1	Indirects	$2,852,436
2	Contingency (12%)	$1,891,402
Subtotal - Indirects		**$4,743,838**
SUB-TOTAL - POST-CLOSURE COSTS		*$20,360,342*
POST-CLOSURE NPV		**$12,621,082**
TOTAL FINANCIAL SECURITY		**$62,701,817**

Worksheet 2 - Year 0 Estimate - Cashflow Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

NPV CALCULATION INPUTS

NOTE: Columns between Years 2052 and 2126 are hidden for printing purposes.

Short-Term Interest Rate (Interim Care/Implementation Costs)

##	Inflation Rate	Years in effect	Start Year	End Year
1	2.0%	100	2018	2118

Input Parameters

Scenario: Year 0 Closure
Current Year: 2018 Year that the estimate is costed
Closure Year 1: 2020
Closure Period: 3 yrs
Post-Closure Year 1: 2023

Long-Term Monitoring/Maintenance NPV Discount Rates

##	Discount Rate	Years in effect	Start Year	End Year
1	2.35%	2	2018	2020
2	2.5%	3	2021	2024
3	2.8%	-	2025	

SCHEDULE

Implementation Schedule

	Area	Closure Year:	INTERIM CARE Year -2	Year -1	CLOSURE Year 1	Year 2	Year 3	POST CLOSURE Year 4	Year 5	Year 6	Year 7	Year 8	Year 9	Year 10	Year 11	Year 12	Year 13	Year 14	Year 15	Year 16	Year 17	Year 18	Year 19	Year 20	Year 21	Year 22	Year 23	Year 24	Year 25	Year 26	Year 27	Year 28	Year 29	Year 30	Year 31	Year 32	Year 33	Year 107	Year 108
		Sum	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2050	2051	2052	2126	2127
Direct Costs	Waste Dumps	100%			60%	40%																																	
	Overburden Dumps	100%					100%																																
	Ore Stockpiles	100%			100%																																		
	Open Pits	100%				50%	50%																																
	Underground Openings	100%					100%																																
	External Tailings Facilities	100%			100%																																		
	Roads	100%					100%																																
	Demolition	100%			10%	75%	15%																																
	Surface Infrastructure	100%			25%	50%	25%																																
	Water Detention Structures	100%				50%	50%																																
	Yards/Laydown Areas	100%			25%	25%	50%																																
	Waste Disposal	100%				12%	75%	2%	2%	2%	6%																														
	Surface Water Conveyance	100%			20%	30%	50%																																
	Water Treatment	100%					86%					14%																												
Indirect	Mobilization-Demobilization	100%			50%		45%					5%																												
	Transport Costs	100%			34%	33%	33%																																
	Road Maintenance	100%			34%	33%	33%																																
	Construction Support	100%			34%	33%	33%																																
	QA and Pro. Mgmt.	100%			34%	33%	33%																																

Monitoring Schedule

		Set Monitoring/C&M Stage:	1 - Interim	1 - Interim	2 - Active	2 - Active	2 - Active	3 - PC 1	3 - PC 1	3 - PC 1	3 - PC 1	3 - PC 1	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2									5 - Perpetual							5 - Perpetual
		Post-Closure Year:	Year -5	Year -4	Year -3	Year -2	Year -1	Year 1	Year 2	Year 3	Year 4	Year 5	Year 6	Year 7	Year 8	Year 9	Year 10	Year 11	Year 12	Year 13	Year 14	Year 15	Year 16	Year 17	Year 18	Year 19	Year 20	Year 21	Year 22	Year 23	Year 24	Year 25	Year 26	Year 27	Year 28	Year 29	Year 30	Year 104	Year 105
		Sum																																					
Planning and permitting	Reclamation Research and Planning	100%	50%	50%																																			
	Technical Studies and Investigations	100%	50%	50%																																			
	Engineering, Design, and Construction Plans	100%	30%	40%	10%	10%	10%																																
	Monitoring and Management Plans	100%	50%	50%																																			
	Permitting	100%	50%	50%																																			
Monitoring	Water Quality Monitoring (surface and groundwater)		1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1									1							1
	Sediment Monitoring		1	1	1	1	1	1		1		1					1					1									1							1
	Biological Monitoring		1	1	1	1	1	1		1		1					1					1									1							1
	Geotechnical Monitoring		1	1	1	1	1	1	1	1	1	1					1					1									1							1
	Revegetation Monitoring				1	1	1	1	1	1	1	1					1					1									1							1
	Annual Inspection Reporting		1	1	1	1	1	1	1	1	1	1					1					1									1							1

Care and Maintenance Schedule

Water Treatment/Management	Active System		x	x	x	x	x	x	x	x	x	x	x	x	x	x	x	x	x	x	x	x									x							x
	Passive System							x	x	x	x	x					x					x									x							x
Reclamation Maintenance	Revegetation, cover repair, etc.							x	x	x	x	x																			x							x
Site Maintenance			x	x	x	x	x	x	x	x	x	x					x					x									x							x
On-site Management			x	x	x	x	x	x	x	x	x	x					x					x									x							x
Transportation Costs			x	x	x	x	x	x	x	x	x	x					x					x									x							x

UNDISCOUNTED CASH FLOW (TO 100 YEARS POST CLOSURE)

Implementation Costs

	Area	Total	% of Total	Check Sum	Year -2	Year -1	Year 1	Year 2	Year 3	Year 4	Year 5	Year 6	Year 7	Year 8	Year 9	Year 10	Year 11	Year 12	Year 13	Year 14	Year 15	Year 16	Year 17	Year 18	Year 19	Year 20	Year 21	Year 22	Year 23	Year 24	Year 25	Year 26	Year 27	Year 28	Year 29	Year 30	Year 31	Year 32	Year 33	Year 107	Year 108
					2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2050	2051	2052	2126	2127
Direct Costs	Waste Dumps	$6,847,702	27%	$6,847,702	$0	$0	$4,108,621	$2,739,081	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Overburden Dumps	$370,112	1%	$370,112	$0	$0	$0	$0	$370,112	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Ore Stockpiles	$1,419,486	6%	$1,419,486	$0	$0	$1,419,486	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Open Pits	$177,927	1%	$177,927	$0	$0	$0	$88,964	$88,964	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Underground Openings	$81,637	0%	$81,637	$0	$0	$0	$0	$81,637	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	External Tailings Facilities	$202,656	1%	$202,656	$0	$0	$202,656	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Roads	$136,050	1%	$136,050	$0	$0	$0	$0	$136,050	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Demolition	$1,976,253	8%	$1,976,253	$0	$0	$197,625	$1,482,189	$296,438	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Surface Infrastructure	$169,527	1%	$169,527	$0	$0	$42,382	$84,763	$42,382	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Water Detention Structures	$786,952	3%	$786,952	$0	$0	$0	$393,476	$393,476	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Yards/Laydown Areas	$758,907	3%	$758,907	$0	$0	$189,727	$189,727	$379,454	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Waste Disposal	$452,923	2%	$452,923	$0	$0	$0	$54,828	$339,692	$11,145	$11,145	$11,145	$24,967	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Surface Water Conveyance	$3,181,261	13%	$3,181,261	$0	$0	$636,252	$954,378	$1,590,630	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Water Treatment	$483,085	2%	$483,085	$0	$0	$0	$0	$413,938	$0	$0	$0	$0	$69,147	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Indirect	Mobilization-Demobilization	$324,077	1%	$324,077	$0	$0	$162,039	$0	$144,406	$0	$0	$0	$0	$17,633	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Transport Costs	$405,599	2%	$405,599	$0	$0	$137,904	$133,848	$133,848	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Site/Road Maintenance	$757,001	3%	$757,001	$0	$0	$257,380	$249,810	$249,810	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Construction Support	$5,054,174	20%	$5,054,174	$0	$0	$1,718,419	$1,667,877	$1,667,877	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	QA and Pro. Mgmt.	$1,668,691	7%	$1,668,691	$0	$0	$567,355	$550,668	$550,668	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Subtotal - Closure Implementation Costs		**$25,254,020**		$25,254,020	**$0**	**$0**	**$9,639,846**	**$8,589,610**	**$6,879,381**	**$11,145**	**$11,145**	**$11,145**	**$24,967**	**$86,780**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

Monitoring Costs

			Total	% of Total	Check Sum	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2050	2051	2052	2126	2127
1	Planning and	Reclamation Research and Planning	$400,000	10%	$400,000	$200,000	$200,000	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2	Permitting	Technical Studies and Investigations	$58,500	1%	$58,500	$29,250	$29,250	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3		Engineering, Design, and Construction Plans	$852,224	21%	$852,224	$255,667	$340,890	$85,222	$85,222	$85,222	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4		Monitoring and Management Plans	$209,158	5%	$209,158	$104,579	$104,579	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5		Permitting	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
1	Monitoring	Water Quality Monitoring	$1,403,014	34%	$1,403,014	$115,295	$115,295	$130,437	$130,437	$130,437	$51,987	$51,987	$51,987	$51,987	$51,987	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$0	$0	$0	$0	$0	$0	$0	$0	$0	$27,664	$0	$0	$0	$0	$0	$27,664
2		Sediment Monitoring	$64,372	2%	$64,372	$3,388	$3,388	$3,388	$3,388	$3,388	$0	$3,388	$0	$3,388	$0	$0	$0	$0	$3,388	$0	$0	$0	$0	$3,388	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$3,388	$0	$0	$0	$0	$0	$3,388
3		Biological Monitoring	$233,587	6%	$233,587	$12,294	$12,294	$12,294	$12,294	$12,294	$0	$12,294	$0	$12,294	$0	$0	$0	$0	$12,294	$0	$0	$0	$0	$12,294	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$12,294	$0	$0	$0	$0	$0	$12,294
4		Geotechnical Monitoring	$379,471	9%	$379,471	$12,409	$12,409	$12,409	$12,409	$12,409	$12,409	$12,409	$12,409	$12,409	$12,409	$0	$0	$0	$23,216	$0	$0	$0	$0	$23,216	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$23,216	$0	$0	$0	$0	$0	$23,216
5		Revegetation Monitoring	$200,071	5%	$200,071	$0	$0	$10,530	$10,530	$10,530	$10,530	$10,530	$10,530	$10,530	$10,530	$0	$0	$0	$10,530	$0	$0	$0	$0	$10,530	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$10,530	$0	$0	$0	$0	$0	$10,530
6		Annual Inspection Reporting	$286,620	7%	$286,620	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$0	$0	$0	$23,900	$0	$0	$0	$0	$23,900	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$23,900	$0	$0	$0	$0	$0	$23,900
Post Closure Costs Subtotal			**$4,087,017**		$4,087,017	**$735,255**	**$820,477**	**$256,653**	**$256,653**	**$256,653**	**$92,981**	**$77,299**	**$92,981**	**$77,299**	**$92,981**	**$27,220**	**$27,220**	**$27,220**	**$27,220**	**$100,548**	**$27,220**	**$27,220**	**$27,220**	**$27,220**	**$100,548**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$100,993**	**$0**	**$0**	**$0**	**$0**	**$0**	**$100,993**

Care and Maintenance Costs

			Total	% of Total	Check Sum	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2050	2051	2052	2126	2127
1	Water Treatment	Active System	$7,981,012	36%	$7,981,012	$706,966.58	$706,967	$706,967	$706,967	$706,967	$706,967	$706,967	$706,967	$706,967	$706,967	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Passive System	$560,239	3%	$560,239	$0	$0	$0	$0	$0	$35,015	$35,015	$35,015	$35,015	$35,015	$0	$0	$0	$0	$35,015	$0	$0	$0	$35,015	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$35,015	$0	$0	$0	$0	$0	$0	$35,015
3	Reclamation Maintenance	(Cover/Reveg Repairs)	$1,834,053	8%	$1,834,053	$0	$0	$0	$0	$0	$0	$366,811	$366,811	$366,811	$366,811	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4	Site Maintenance		$2,478,551	11%	$2,478,551	$359,187	$359,187	$30,000	$30,000	$30,000	$99,414	$99,414	$99,414	$99,414	$99,414	$0	$0	$0	$0	$99,414	$0	$0	$0	$0	$99,414	$0	$0	$0	$0	$0	$0	$0	$0	$0	$108,253	$0	$0	$0	$0	$0	$108,253
5	On-site Management		$7,978,119	36%	$7,978,119	$1,124,816	$1,124,816	$310,135	$310,135	$310,135	$606,140	$606,140	$606,140	$606,140	$606,140	$0	$0	$0	$0	$79,830	$0	$0	$0	$0	$79,830	$0	$0	$0	$0	$0	$0	$0	$0	$0	$178,636	$0	$0	$0	$0	$0	$178,636
6	Transportation Costs		$1,297,725	6%	$1,297,725	$151,714	$151,714	$0	$0	$0	$64,628	$64,628	$64,628	$64,628	$64,628	$0	$0	$0	$0	$20,832	$0	$0	$0	$0	$20,832	$0	$0	$0	$0	$0	$0	$0	$0	$0	$69,943	$0	$0	$0	$0	$0	$69,943
Post Closure Costs Subtotal			**$22,129,700**		$22,129,700	**$2,342,683**	**$2,342,683**	**$1,047,102**	**$1,047,102**	**$1,047,102**	**$1,878,975**	**$1,878,975**	**$1,878,975**	**$1,878,975**	**$1,878,975**	**$91,135**	**$91,135**	**$91,135**	**$91,135**	**$326,226**	**$91,135**	**$91,135**	**$91,135**	**$91,135**	**$326,226**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$391,847**	**$0**	**$0**	**$0**	**$0**	**$0**	**$391,847**

Other Add-on Costs

			Total	% of Total	Check Sum	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2050	2051	2052	2126	2127
1	Contingency	12% % of above costs	$6,176,488		$6,176,488	$369,353	$379,579	$1,313,232	$1,187,204	$981,976	$237,972	$236,090	$237,972	$237,749	$247,048	$14,203	$14,203	$14,203	$14,203	$51,213	$14,203	$14,203	$14,203	$14,203	$51,213	$0	$0	$0	$0	$0	$0	$0	$0	$0	$59,141	$0	$0	$0	$0	$0	$59,141
2	Worker's compensation	0% (Included in labour rates)	$0		$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3	Insurance	1.5% % of above labour costs	$220,355.80		$220,356	$13,177	$13,542	$46,852	$42,355	$35,034	$8,490	$8,423	$8,490	$8,482	$8,814	$507	$507	$507	$507	$1,827	$507	$507	$507	$507	$1,827	$0	$0	$0	$0	$0	$0	$0	$0	$0	$2,110	$0	$0	$0	$0	$0	$2,110
4	Bonding	3% % of Implem. & C&M costs	$1,421,511.59		$1,421,512	$70,280.50	$70,280	$320,608	$289,101	$237,794	$56,704	$56,704	$56,704	$57,118	$58,973	$2,734	$2,734	$2,734	$2,734	$9,787	$2,734	$2,734	$2,734	$2,734	$9,787	$0	$0	$0	$0	$0	$0	$0	$0	$0	$11,755	$0	$0	$0	$0	$0	$11,755
5	Contractor Profit	10% % of above costs	$5,147,074		$5,147,074	$307,794	$316,316	$1,094,360	$989,337	$818,314	$198,310	$196,742	$198,310	$198,124	$205,874	$11,835	$11,835	$11,835	$11,835	$42,677	$11,835	$11,835	$11,835	$11,835	$42,677	$0	$0	$0	$0	$0	$0	$0	$0	$0	$49,284	$0	$0	$0	$0	$0	$49,284
6	Contract Administration	5% % of above costs	$2,573,537		$2,573,537	$153,897	$158,158	$547,180	$494,668	$409,157	$99,155	$98,371	$99,155	$99,062	$102,937	$5,918	$5,918	$5,918	$5,918	$21,339	$5,918	$5,918	$5,918	$5,918	$21,339	$0	$0	$0	$0	$0	$0	$0	$0	$0	$24,642	$0	$0	$0	$0	$0	$24,642
Other Add-On Costs Subtotal			**$15,538,966**		$15,538,966	**$914,501**	**$937,876**	**$3,322,232**	**$3,002,665**	**$2,482,275**	**$600,631**	**$596,330**	**$600,631**	**$600,535**	**$623,645**	**$35,196**	**$35,196**	**$35,196**	**$35,196**	**$126,843**	**$35,196**	**$35,196**	**$35,196**	**$35,196**	**$126,843**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$146,932**	**$0**	**$0**	**$0**	**$0**	**$0**	**$146,932**
TOTAL		**$67,009,703**		$67,009,703	$3,992,439	$4,101,036	$14,265,834	$12,896,030	$10,665,411	$2,583,731	$2,563,748	$2,583,731	$2,581,775	$2,682,381	$153,551	$153,551	$153,551	$153,551	$553,617	$153,551	$153,551	$153,551	$153,551	$553,617	$0	$0	$0	$0	$0	$0	$0	$0	$0	$639,772	$0	$0	$0	$0	$0	$639,772

INFLATION FACTORED INTERIM CARE AND CLOSURE IMPLEMENTATION COSTS

	Area	Present Value	% of Total	Future Value	Year -2 2018	Year -1 2019	Year 1 2020	Year 2 2021	Year 3 2022	Year 4 2023	Year 5 2024	Year 6 2025	Year 7 2026	Year 8 2027	Year 9 2028	Year 10 2029	Year 11 2030	Year 12 2031	Year 13 2032	Year 14 2033	Year 15 2034	Year 16 2035	Year 17 2036	Year 18 2037	Year 19 2038	Year 20 2039	Year 21 2040	Year 22 2041	Year 23 2042	Year 24 2043	Year 25 2044	Year 26 2045	Year 27 2046	Year 28 2047	Year 29 2048	Year 30 2049	Year 31 2050	Year 32 2051	Year 33 2052	Year 107 2126	Year 108 2127
Direct Costs	Waste Dumps	$6,847,702	27%	$7,181,344	$0	$0	$4,274,610	$2,906,735	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Overburden Dumps	$370,112	1%	$400,621	$0	$0	$0	$0	$400,621	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Ore Stockpiles	$1,419,486	6%	$1,476,833	$0	$0	$1,476,833	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Open Pits	$177,927	1%	$190,706	$0	$0	$0	$94,409	$96,297	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Underground Openings	$81,637	0%	$88,367	$0	$0	$0	$0	$88,367	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	External Tailings Facilities	$202,656	1%	$210,844	$0	$0	$210,844	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Roads	$136,050	1%	$147,265	$0	$0	$0	$0	$147,265	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Demolition	$1,976,253	8%	$2,099,395	$0	$0	$205,609	$1,572,911	$320,874	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Surface Infrastructure	$169,527	1%	$179,921	$0	$0	$44,094	$89,952	$45,875	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Water Detention Structures	$786,952	3%	$843,470	$0	$0	$0	$417,560	$425,911	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Yards/Laydown Areas	$758,907	3%	$809,464	$0	$0	$197,392	$201,340	$410,733	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Waste Disposal	$452,923	2%	$492,790	$0	$0	$0	$58,184	$367,694	$12,305	$12,551	$12,802	$29,253	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Surface Water Conveyance	$3,181,261	13%	$3,396,500	$0	$0	$661,957	$1,012,794	$1,721,750	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Water Treatment	$483,085	2%	$530,697	$0	$0	$0	$0	$448,060	$0	$0	$0	$0	$82,637	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Indirect	Mobilization-Demobilization	$324,077	1%	$345,967	$0	$0	$168,585	$0	$156,309	$0	$0	$0	$0	$21,073	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Transport Costs	$405,599	2%	$430,396	$0	$0	$143,475	$142,040	$144,881	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Road Maintenance	$757,001	3%	$803,282	$0	$0	$267,779	$265,101	$270,403	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Construction Support	$5,054,174	20%	$5,363,172	$0	$0	$1,787,843	$1,769,965	$1,805,364	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	QA and Pro. Mgmt.	$1,668,691	7%	$1,770,711	$0	$0	$590,276	$584,373	$596,061	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Subtotal - Closure Implementation Costs		**$25,254,020**		**$26,761,744**	**$0**	**$0**	**$10,029,296**	**$9,115,363**	**$7,446,463**	**$12,305**	**$12,551**	**$12,802**	**$29,253**	**$103,710**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

Monitoring Costs

			Present Value	% of Total	Future Value	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2050	2051	2052	2126	2127
1	Planning and permitting	Reclamation Research and Planning	$400,000	10%	$404,000	$200,000	$204,000	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Technical Studies and Investigations	$58,500	1%	$59,085	$29,250	$29,835	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3		Engineering, Design, and Construction Plans	$852,224	21%	$874,726	$255,667	$347,707	$88,665	$90,439	$92,247	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4		Monitoring and Management Plans	$209,158	5%	$211,250	$104,579	$106,671	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5		Permitting	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
1	Monitoring	Water Quality Monitoring	$621,903	15%	$648,215	$115,295	$117,601	$135,707	$138,421	$141,190	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Sediment Monitoring	$16,940	0%	$17,631	$3,388	$3,456	$3,525	$3,595	$3,667	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3		Biological Monitoring	$61,470	2%	$63,979	$12,294	$12,540	$12,791	$13,047	$13,307	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4		Geotechnical Monitoring	$62,045	2%	$64,577	$12,409	$12,657	$12,910	$13,169	$13,432	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5		Revegetation Monitoring	$31,590	1%	$33,528	$0	$0	$10,955	$11,175	$11,398	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
6		Annual Inspection Reporting	$11,860	0%	$12,344	$2,372	$2,419	$2,468	$2,517	$2,568	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Post Closure Costs Subtotal			**$2,325,691**		**$2,389,335**	**$735,255**	**$836,887**	**$267,022**	**$272,362**	**$277,809**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

Care and Maintenance Costs

			Present Value	% of Total	Future Value	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2050	2051	2052	2126	2127
1	Water Treatment	Active System	$3,534,833	16%	$3,679,082	$706,967	$721,106	$735,528	$750,239	$765,243	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Passive System	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3	Reclamation Maintenance	(Cover/Reveg Repairs)	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4	Site Maintenance		$808,374	4%	$821,079	$359,187	$366,371	$31,212	$31,836	$32,473	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5	On-site Management		$3,180,037	14%	$3,259,611	$1,124,816	$1,147,312	$322,665	$329,118	$335,701	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
6	Transportation Costs		$303,428	1%	$306,462	$151,714	$154,748	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Post Closure Costs Subtotal			**$7,826,673**		$8,066,235	**$2,342,683**	**$2,389,537**	**$1,089,405**	**$1,111,193**	**$1,133,417**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

Other Add-on Costs

			Present Value	% of Total	Future Value	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2050	2051	2052	2126	2127
1	Contingency	12% % of above costs	$4,231,344		$4,445,603	$369,353	$387,171	$1,366,287	$1,259,870	$1,062,923	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2	Worker's compensation	0% (Included in labour rates)	$0		$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3	Insurance	2% % of above labour costs	$150,960		$158,604	$13,177	$13,813	$48,744	$44,948	$37,921	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4	Bonding	3% % of Implem. & C&M costs	$988,065		$1,039,721	$70,280	$71,686	$333,561	$306,797	$257,396	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5	Contractor Profit	10% % of above costs	$3,526,120		$3,704,669	$307,794	$322,642	$1,138,572	$1,049,892	$885,769	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
6	Contract Administration	5% % of above costs	$1,763,060		$1,852,335	$153,897	$161,321	$569,286	$524,946	$442,884	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Other Add-On Costs Subtotal			**$10,659,549**		$11,200,931	**$914,501**	**$956,633**	**$3,456,451**	**$3,186,452**	**$2,686,894**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

TOTAL			**$46,065,933**		**$48,418,245**	**$3,992,439**	**$4,183,057**	**$14,842,173**	**$13,685,371**	**$11,544,584**	**$12,305**	**$12,551**	**$12,802**	**$29,253**	**$103,710**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

NET PRESENT VALUE CALCULATIONS - LONG TERM MONITORING AND MAINTENANCE

Monitoring Costs

			Undiscounted	% of Total	NPV - 2018	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2050	2051	2052	2126	2127
1	Monitoring	Water Quality Monitoring	$781,111	19%	$463,795						$46,084	$46,084	$44,829	$43,608	$42,420	$21,605	$21,017	$20,444	$19,888	$19,346	$18,819	$18,306	$17,808	$17,323	$16,851	$0	$0	$0	$0	$0	$0	$0	$0	$0	$12,994	$0	$0	$0	$0	$0	$0	$1,427
2		Sediment Monitoring	$47,432	1%	$19,240						$3,003	$0	$2,921	$0	$2,765	$0	$0	$0	$0	$2,408	$0	$0	$0	$0	$2,097	$0	$0	$0	$0	$0	$0	$0	$0	$0	$1,591	$0	$0	$0	$0	$0	$0	$175
3		Biological Monitoring	$172,117	4%	$69,816						$10,898	$0	$10,601	$0	$10,032	$0	$0	$0	$0	$8,738	$0	$0	$0	$0	$7,611	$0	$0	$0	$0	$0	$0	$0	$0	$0	$5,774	$0	$0	$0	$0	$0	$0	$634
4		Geotechnical Monitoring	$317,426	8%	$125,534						$11,000	$11,000	$10,700	$10,409	$10,125	$0	$0	$0	$0	$16,501	$0	$0	$0	$0	$14,373	$0	$0	$0	$0	$0	$0	$0	$0	$0	$10,905	$0	$0	$0	$0	$0	$0	$1,197
5		Revegetation Monitoring	$168,481	4%	$77,966						$9,334	$9,334	$9,080	$8,833	$8,592	$0	$0	$0	$0	$7,484	$0	$0	$0	$0	$6,519	$0	$0	$0	$0	$0	$0	$0	$0	$0	$4,946	$0	$0	$0	$0	$0	$0	$543
6		Annual Inspection Reporting	$274,760	7%	$84,604						$2,103	$2,103	$2,045	$1,990	$1,935	$0	$0	$0	$0	$16,987	$0	$0	$0	$0	$14,796	$0	$0	$0	$0	$0	$0	$0	$0	$0	$11,226	$0	$0	$0	$0	$0	$0	$1,232
Post Closure Costs Subtotal			**$1,761,326**		**$840,955**	**$0**	**$0**	**$0**	**$0**	**$0**	**$82,422**	**$68,521**	**$80,177**	**$64,839**	**$75,869**	**$21,605**	**$21,017**	**$20,444**	**$19,888**	**$71,463**	**$18,819**	**$18,306**	**$17,808**	**$17,323**	**$62,247**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$47,435**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$5,208**

Care and Maintenance Costs

			Undiscounted	% of Total	NPV - 2018	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2050	2051	2052	2126	2127
1	Water Treatment	Active System	$4,446,180	20%	$3,673,728						$626,688	$626,688	$609,619	$593,015	$576,862	$72,338	$70,367	$68,451	$66,586	$64,773	$63,008	$61,292	$59,623	$57,999	$56,419	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Passive System	$560,239	3%	$259,255						$31,039	$31,039	$30,193	$29,371	$28,571	$0	$0	$0	$0	$24,886	$0	$0	$0	$0	$21,677	$0	$0	$0	$0	$0	$0	$0	$0	$0	$16,446	$0	$0	$0	$0	$0	$0	$1,806
3	Reclamation Maintenance	(Cover/Reveg Repairs)	$1,834,053	8%	$1,573,611						$325,158	$325,158	$316,302	$307,687	$299,306	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4	Site Maintenance		$1,670,177	8%	$751,847						$88,125	$88,125	$85,725	$83,390	$81,119	$0	$0	$0	$0	$70,657	$0	$0	$0	$0	$61,545	$0	$0	$0	$0	$0	$0	$0	$0	$0	$50,845	$0	$0	$0	$0	$0	$0	$5,582
5	On-site Management		$4,798,081	22%	$3,025,236						$537,311	$537,311	$522,676	$508,440	$494,591	$0	$0	$0	$0	$56,738	$0	$0	$0	$0	$49,420	$0	$0	$0	$0	$0	$0	$0	$0	$0	$83,903	$0	$0	$0	$0	$0	$0	$9,211
6	Transportation Costs		$994,297	4%	$429,761						$57,290	$57,290	$55,729	$54,211	$52,735	$0	$0	$0	$0	$14,806	$0	$0	$0	$0	$12,897	$0	$0	$0	$0	$0	$0	$0	$0	$0	$32,852	$0	$0	$0	$0	$0	$0	$3,607
Post Closure Costs Subtotal			**$14,303,027**		$9,713,438	**$0**	**$0**	**$0**	**$0**	**$0**	**$1,665,611**	**$1,665,611**	**$1,620,245**	**$1,576,113**	**$1,533,184**	**$72,338**	**$70,367**	**$68,451**	**$66,586**	**$231,860**	**$63,008**	**$61,292**	**$59,623**	**$57,999**	**$201,958**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$184,047**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$20,206**

Other Add-on Costs

			Undiscounted	% of Total	NPV - 2018	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2050	2051	2052	2126	2127
1	Contingency	12% % of above costs	$1,945,144		$1,281,062						$210,950	$209,281	$205,204	$199,427	$201,584	$11,273	$10,966	$10,667	$10,377	$36,399	$9,819	$9,552	$9,292	$9,039	$31,705	$0	$0	$0	$0	$0	$0	$0	$0	$0	$27,778	$0	$0	$0	$0	$0	$0	$3,050
2	Worker's compensation	0% (Included in labour rates)	$0		$0						$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3	Insurance	2% % of above labour costs	$69,396		$45,704						$7,526	$7,466	$7,321	$7,115	$7,192	$402	$391	$381	$370	$1,299	$350	$341	$331	$322	$1,131	$0	$0	$0	$0	$0	$0	$0	$0	$0	$991	$0	$0	$0	$0	$0	$0	$109
4	Bonding	3% % of Implem. & C&M costs	$433,446		$295,037						$50,265	$50,265	$48,896	$47,912	$48,120	$2,170	$2,111	$2,054	$1,998	$6,956	$1,890	$1,839	$1,789	$1,740	$6,059	$0	$0	$0	$0	$0	$0	$0	$0	$0	$5,521	$0	$0	$0	$0	$0	$0	$606
5	Contractor Profit	10% % of above costs	$1,620,954		$1,067,552						$175,791	$174,401	$171,003	$166,190	$167,986	$9,394	$9,138	$8,890	$8,647	$30,332	$8,183	$7,960	$7,743	$7,532	$26,420	$0	$0	$0	$0	$0	$0	$0	$0	$0	$23,148	$0	$0	$0	$0	$0	$0	$2,541
6	Contract Administration	5% % of above costs	$810,477		$533,776						$87,896	$87,201	$85,502	$83,095	$83,993	$4,697	$4,569	$4,445	$4,324	$15,166	$4,091	$3,980	$3,872	$3,766	$13,210	$0	$0	$0	$0	$0	$0	$0	$0	$0	$11,574	$0	$0	$0	$0	$0	$0	$1,271
Other Add-On Costs Subtotal			**$4,879,417**		$3,223,130	**$0**	**$0**	**$0**	**$0**	**$0**	**$532,427**	**$528,614**	**$517,925**	**$503,738**	**$508,875**	**$27,937**	**$27,176**	**$26,436**	**$25,716**	**$90,152**	**$24,334**	**$23,671**	**$23,026**	**$22,399**	**$78,525**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$69,012**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$7,577**

TOTAL			**$20,943,770**		**$13,777,523**	**$0**	**$0**	**$0**	**$0**	**$0**	**$2,280,461**	**$2,262,747**	**$2,218,347**	**$2,144,691**	**$2,117,928**	**$121,880**	**$118,560**	**$115,331**	**$112,190**	**$393,475**	**$106,161**	**$103,270

Worksheet 3 - Year 0 Estimate - Annual C&M Costs

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

Summary of Annual Care and Maintenance Costs

		Annual Cost Per Phase							Number of Years				
		1 - Interim	2 - Active	3 - PC 1	4 - PC 2	5 - Perpetual	n/a	Total Cost	1 - Interim	2 - Active	3 - PC 1	4 - PC 2	5 - Perpetual
1	Active Water Treatment System	$706,967	$706,967	$706,967	$91,135	$0	$0	$7,981,012	2	3	5	10	0
2	Passive Treatment System	$0	$0	$35,015	$35,015	$35,015	$0	$560,239	2	3	5	2	9
3	Reclamation Maintenance	$0	$0	$366,811	$0	$0	$0	$1,834,053	2	3	5	2	9
4	Site Maintenance	$359,187	$30,000	$99,414	$99,414	$108,253	$0	$2,478,551	2	3	5	2	9
5	On-site Management	$1,124,816	$310,135	$606,140	$79,830	$178,636	$0	$7,978,119	2	3	5	2	9
6	Transportation Costs	$151,714	$0	$64,628	$20,832	$69,943	$0	$1,297,725	2	3	5	2	9

NOTES:
1. This table is used as a check to the NPV calculation spreadsheet.

Phase 1 - Annual Care & Maintenance Costs - Prior to Closure

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S1.1	**Water Treatment**															**$706,967**	
S1.1.1	*Active Treatment System*														$706,967		
S1.1.1.1		Operation and maintenance		1	yr						$511,807	$65,194	$38,831	$615,832			See details on 'Water Treatment' worksheet.
S1.1.1.2		Capital Replacement	annual allowance	1	yr						$86,932	$4,203	$0	$91,135			See details on 'Water Treatment' worksheet.
S1.2	**Onsite Management**															**$1,124,816**	
S1.2.1	*Field Support Staff*														$761,189		
S1.2.1.1		Mine Manager		4	months	360	site working hours per month			$23,408				$81,930			(Notes: staffing costs include cross-shifts)
S1.2.1.2		Office/Camp manager		6	months	360	site working hours per month			$22,267				$133,604			
S1.2.1.3		Administrative Assistants/HR/Accounting		4	months	360	site working hours per month			$16,600				$58,098			
S1.2.1.4		Water Treatment Staff Operator		5	months	360	site working hours per month			$18,912				$94,560			
S1.2.1.5		Environmental/Safety Manager		4	months	360	site working hours per month			$22,267				$89,069			
S1.2.1.6		Environmental Technicians		7	months	360	site working hours per month			$16,046				$104,301			
S1.2.1.7		Mechanic		2	months	360	site working hours per month			$17,769				$35,538			
S1.2.1.8		Tradesmen		2	months	360	site working hours per month			$18,912				$42,552			
S1.2.1.9		General Labour/helpers		10	months	360	site working hours per month			$12,793				$121,536			
S1.2.2	*Field support Vehicles*														$131,186		
S1.2.2.1		Pick-up trucks (3 required)		12	months					$4,446				$53,350			
S1.2.2.2		Mechanic service vehicle		2	months					$11,803				$23,607			
S1.2.2.3		Emergency transport vehicle		12	months					$4,519				$54,229			
S1.2.3	*Field Support Equipment/Supplies*														$232,440		
S1.2.3.1		Office supplies		12	months					$1,000				$12,000			
S1.2.3.2		Communications		12	months					$1,000				$12,000			
S1.2.3.3		Misc. supplies		12	months					$500				$6,000			
S1.2.3		Camp Operation		1,995	man-days					$80				$159,600			
S1.2.3		Power, heat, fuel		12	months					$3,570				$42,840			
S1.3	**Site Maintenance**														$359,187	**$359,187**	
S1.3.1		Equipment Maintenance		1	ls		$20,000							$20,000			Labour included in field support staff
S1.3.2		Road Maintenance	Grader (assume 80 hrs/month)	1	ls	960		$44	$106	$150	$0	$42,612	$101,443	$144,056			Operator cost included in Field support staff
S1.3.3		Earthwork Repair allowance (assume 40 hrs per month)		1	ls	480	$0	$132	$344	$344			$165,132	$165,132			Task Code C.2.14
S1.3.4		Sundry equipment/consumables allowance		1	ls					$30,000				$30,000			Allowance for pump maintenance/fuel etc.
S1.4	**Transportation Costs**														$151,714	**$151,714**	
S1.4.1		Barge Operations		5	months					$10,832				$54,162			
S1.4.2		Staffing Bus trips during barge operation period		20	ea					$678				$13,552			One per week
S1.4.3		Air transport and airstrip operations		8	flights					$3,000				$24,000			One flight per week
S1.4.4		Helicopter transport		20	flights					$3,000				$60,000			One flight per week
TOTAL														**$2,342,683**	**$2,342,683**	**$2,342,683**	

Phase 2 - Annual Care & Maintenance Costs During Active Closure

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S2.1	**Water Treatment**															**$706,967**	
S2.1.1	*Active Treatment System*														$706,967		
S2.1.1.1		Operation and maintenance		1	yr						$511,807	$65,194	$38,831	$615,832			See details on 'Water Treatment' worksheet.
S2.1.1.2		Capital Replacement	annual allowance	1	yr						$86,932	$4,203	$0	$91,135			See details on 'Water Treatment' worksheet.
S2.2	**Onsite Management**															**$310,135**	
S2.2.1	*Field Support Staff*														$247,815		
S2.2.1.1		Water Treatment Staff Operator		5	months	360	site working hours per month			$18,912				$94,560			
S2.2.1.2		Environmental/Safety Manager		4	months	360	site working hours per month			$22,267				$89,069			
S2.2.1.3		Environmental Technicians		4	months	360	site working hours per month			$16,046				$64,185			
S2.2.2	*Field support Vehicles*														$31,121		
S2.2.2.1		Pick-up trucks (1 required)		7	months					$4,446				$31,121			
S2.2.3	*Field Support Equipment/Supplies*														$31,200		
S2.2.3.1		Camp Operation		390	man-days					$80				$31,200			
S2.3	**Site Maintenance**														$30,000	**$30,000**	
S2.3.1		Pumping equipment/consumables allowance		1	ls					$30,000				$30,000			Allowance for pump maintenance/fuel etc.
S2.4	**Transportation Costs**														$0	**$0**	
S2.4.1		Barge Operations		0	months					$10,832				$0			Included in construction indirect costs
TOTAL														**$1,047,102**	**$1,047,102**	**$1,047,102**	

Phase 3 - Annual Care & Maintenance Costs - Post Closure I

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours or Cost Code Ref.	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S3.1	**Water Treatment**															**$741,982**	
S3.1.1	*Active Treatment System*														$706,967		
S3.1.1.1		Operation and maintenance		1	yr						$511,807	$65,194	$38,831	$615,832			See details on 'Water Treatment' worksheet.
S3.1.1.2		Capital Replacement	annual allowance	1	yr						$86,932	$4,203	$0	$91,135			See details on 'Water Treatment' worksheet.
S3.1.1	*Passive Treatment System*														$35,015		
S3.1.1.1		Operation and maintenance		1	yr						$10,000	$11,986	$13,029	$35,015			See details on 'Water Treatment' worksheet.
S3.2	**Onsite Management**															**$606,140**	
S3.2.1	*Field Support Staff*														$377,100		
S3.2.1.1		Mine Manager		3	months	360	site working hours per month			$23,408				$58,521			(Notes: staffing costs include cross-shifts)
S3.2.1.2		Office/Camp manager		3	months	360	site working hours per month			$22,267				$66,802			
S3.2.1.3		Administrative Assistants/HR/Accounting		1	months	360	site working hours per month			$16,600				$20,749			
S3.2.1.4		Water Treatment Staff Operator		2	months	360	site working hours per month			$18,912				$33,096			
S3.2.1.5		Environmental/Safety Manager		3	months	360	site working hours per month			$22,267				$55,668			
S3.2.1.6		Environmental Technicians		3	months	360	site working hours per month			$16,046				$40,116			
S3.2.1.7		Mechanic		2	months	360	site working hours per month			$17,769				$35,538			
S3.2.1.8		Tradesmen		2	months	360	site working hours per month			$18,912				$37,824			
S3.2.1.9		General Labour/helpers		2	months	360	site working hours per month			$12,793				$28,785			
S3.2.2	*Field support Vehicles*														$112,890		
S3.2.2.1		Pick-up trucks (3 required)		15	months					$4,446				$66,688			
S3.2.2.2		Mechanic service vehicle		2	months					$11,803				$23,607			
S3.2.2.3		Emergency transport vehicle		5	months					$4,519				$22,596			
S3.2.3	*Field Support Equipment/Supplies*														$116,150		
S3.2.3.1		Office supplies		5	months					$1,000				$5,000			
S3.2.3.2		Communications		5	months					$1,000				$5,000			
S3.2.3.3		Misc. supplies		5	months					$500				$2,500			
S3.2.3		Camp Operation		1,073	man-days					$80				$85,800			
S3.2.3		Power, heat, fuel		5	months					$3,570				$17,850			
S3.3	**Reclamation Maintenance**															**$366,811**	
S3.3.1	*Cover repairs*														$320,441		
S3.3.1.1		Cover Repair allowance	Load, haul, dump, spread (spoil lo	2%	of total	1,336,364	m3 of cover required replacement at:			$5.31	/m3			$142,044			Assumes 10% of area reseeded over 5 yr PC1 perio
S3.3.1.2			Seed/Fertilize: broadcast seeding	67	ha					$2,669.89	/ha			$178,397			
S3.3.2.1	*Revegetation*														$46,370		
S3.3.2.2		Reseeding allowance	Seed/Fertilize: broadcast seeding	5%	of total	347	ha requires reseeding at			$2,669.89	/ha			$46,370			Assumes 25% of area reseeded over 5 yr PC1 perio
S3.3	**Site Maintenance**														$99,414	**$99,414**	
S3.3.1		Equipment Maintenance		1	ls		$20,000							$20,000			Labour included in field support staff
S3.3.2		Road Maintenance	Grader (assume 40 hrs/month)	1	ls	200		$44	$106	$150	$0	$8,878	$21,134	$30,012			Operator cost included in Field support staff
S3.3.3		Earthwork Repair allowance (assume 20 hrs per month)		1	ls	100	$0	$132	$344	$344			$34,402	$34,402			Task Code C.2.14
S3.3.4		Sundry equipment/consumables allowance		1	ls					$15,000				$15,000			Allowance for pump maintenance/fuel etc.
S3.4	**Transportation Costs**														$64,628	**$64,628**	
S3.4.1		Barge Operations		3	months					$10,832				$32,497			
S3.4.2		Staffing Bus trips during barge operation period		12	ea					$678				$8,131			One per week
S3.4.3		Air transport and airstrip operations		8	flights					$3,000				$24,000			One flights per week
TOTAL														**$1,878,975**	**$1,878,975**	**$1,878,975**	

Phase 4 - Annual Care & Maintenance Costs - Post Closure II

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours or Cost Code Ref.	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S4.1	**Water Treatment**															**$126,150**	
S4.1.1	*Active Treatment System*														$91,135		
S4.1.1.2		Capital Replacement	annual allowance	1	yr						$86,932	$4,203	$0	$91,135			See details on 'Water Treatment' worksheet.
S4.1.1	*Passive Treatment System*														$35,015		
S4.1.1.1		Operation and maintenance		1	yr						$10,000	$11,986	$13,029	$35,015			See details on 'Water Treatment' worksheet.
S4.2	**Onsite Management**															**$79,830**	
S4.2.1	*Field Support Staff*														$44,847		
S4.2.1.1		Project Manager		0.5	months	360	site working hours per month			$23,408				$11,704			(Notes: staffing costs include cross-shifts)
S4.2.1.2		Passive Treatment specialist		0.5	months	360	site working hours per month			$42,120				$21,060			
S4.2.1.3		Mechanic		0.5	months	360	site working hours per month			$17,769				$8,885			
S4.2.1.4		General Labour/helpers		0.3	months	360	site working hours per month			$12,793				$3,198			
S4.2.2	*Field support Vehicles*														$19,313		
S4.2.2.1		Pick-up trucks (2 required)		2.0	months					$4,446				$8,892			
S4.2.2.2		Mechanic service vehicle		0.5	months					$11,803				$5,902			
S4.2.2.3		Emergency transport vehicle		1.0	months					$4,519				$4,519			
S4.2.3	*Field Support Equipment/Supplies*														$15,670		
S4.2.3.1		Office supplies		1	months					$1,000				$1,000			
S4.2.3.2		Communications		1	months					$1,000				$1,000			
S4.2.3.3		Misc. supplies		1	months					$500				$500			
S4.2.3		Camp Operation		120	man-days					$80				$9,600			
S4.2.3		Power, heat, fuel		1	months					$3,570				$3,570			
S4.3	**Reclamation Maintenance**															**$0**	
S4.3.1	*Cover repairs*														$0		
S4.3.2	*Revegetation*														$0		
S4.3	**Site Maintenance**														$29,882	**$29,882**	
S4.3.1		Equipment Maintenance (parts, supplies)		1	ls		$10,000							$10,000			Labour included in field support staff
S4.3.2		Road Maintenance	Grader (assume 20 hrs/month)	1	ls	20		$44	$106	$150	$0	$888	$2,113	$3,001			Operator cost included in Field support staff
S4.3.3		Earthwork Repair allowance (assume 20 hrs per month)		1	ls	20	$0	$132	$344	$344			$6,880	$6,880			Task Code C.2.14
S4.3.4		Sundry equipment/consumables allowance		1	ls					$10,000				$10,000			Allowance for pump maintenance/fuel etc.
S4.4	**Transportation Costs**														$20,832	**$20,832**	
S4.4.1		Barge Operations		1	months					$10,832				$10,832			
S4.4.2		Staffing Bus trips during barge operation period		0	ea					$678				$0			Pick-up trucks used (time included in labour rate)
S4.4.3		Barge mob/demob		1	LS					$10,000				$10,000			
TOTAL														**$256,693**	**$256,693**	**$256,693**	

Phase 5 - Perpetual Care & Maintenance Event Year Annual Costs

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours or Cost Code Ref.	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S5.1	**Water Treatment**															**$35,015**	
S5.1.1	*Passive Treatment System*														$35,015		
S5.1.1.1		Operation and maintenance		1	yr						$10,000	$11,986	$13,029	$35,015			See details on 'Water Treatment' worksheet.
S5.2	**Onsite Management**															**$178,636**	
S5.2.1	*Field Support Staff*														$129,326		
S5.2.1.1		Project Manager		1.0	months	360	site working hours per month			$23,408				$23,408			(Notes: staffing costs include cross-shifts)
S5.2.1.2		Passive Treatment specialist		1.0	months	360	site working hours per month			$42,120				$42,120			
S5.2.1.3		Site Engineer		1.0	months	360	site working hours per month			$42,120				$42,120			
S5.2.1.4		HD mechanic		0.5	months	360	site working hours per month			$17,769				$8,885			
S5.2.1.5		General Labour/helpers		1.0	months	360	site working hours per month			$12,793				$12,793			
S5.2.2	*Field support Vehicles*														$19,239		
S5.2.2.1		Pick-up trucks (3 required)		3	months					$4,446				$13,338			
S5.2.2.2		Mechanic service vehicle		1	months					$11,803				$5,902			
S5.2.3	*Field Support Equipment/Supplies*														$30,070		
S5.2.3.1		Office supplies		1	months					$1,000				$1,000			
S5.2.3.2		Communications		1	months					$1,000				$1,000			
S5.2.3.3		Misc. supplies		1	months					$500				$500			
S5.2.3.4		Camp Operation		300	man-days					$80				$24,000			
S5.2.3.5		Power, heat, fuel		1	months					$3,570				$3,570			
S5.3	**Reclamation Maintenance**															**$0**	
S5.4	**Site Maintenance**														$108,253	**$108,253**	
S5.4.1		Grader Maintenance	Grader (assume 2 weeks, 10hrs/d	1	ls	140		$44	$106	$150	$0	$6,214	$14,794	$21,008			Operator cost included in Field support staff
S5.4.2		Earthwork Repair allowance (assume 3 weeks, 10hrs/day)		1	ls	210	$0	$132	$344	$344			$72,245	$72,245			Task Code C.2.14
S5.4.3		Big Creek Bridge - Capital replacement allowance		1	ls					$15,000				$15,000			Allowance for supplies, pumping, etc.
S5.4.4		Sundry allowance		1	ls					$2,500				$2,500			Allowance for supplies, pumping, etc.
S5.5	**Transportation Costs**														$69,943	**$69,943**	
S5.5.1		Barge Operations		1	months					$10,832				$10,832			
S5.5.2		Barge mobilization/demob and set up		1	ls					$10,000				$10,000			assumed
S5.5.3		Camp Mob/demob		1	ls					$10,000				$10,000			assumed
S5.5.4		Equipment mobilization and demobilization		2	ls					$19,555				$39,110.91			See details on 'mob-demob 'worksheet
TOTAL														**$394,347**	**$391,847**	**$391,847**	

Worksheet 4 - Year 0 Estimate - Implementation Costs

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
DIRECT COSTS																					
A1	**Waste Dumps**																			**$6,847,702**	
A1.1	*Main Waste Dump (inc. MWDE)*																		$776,082		Does not include MWD Wrap
A1.1.1.1		Regrade	Flat areas: Re-grade to form tertiary drainage catchments	187	hrs	C.2.13	187	$44.39	$8,303	$194.68	$36,415	$67.01	13,056	$12,534	$0.00	$0	$306.07	$57,252			
A1.1.1.2			Slopes: Re-grade to 3H:1V	336	hrs	C.2.13	336	$44.39	$14,916	$194.68	$65,421	$67.01	23,456	$22,518	$0.00	$0	$306.07	$102,855			
A1.1.2.1		Cover	Flat area cover material: Load, haul, dump spread	22,513	Cm3	R.050	420.0	$0.82	$18,498	$2.67	$60,112	$0.88	20,606	$19,782	$0.00	$0	$4.37	$98,392			
A1.1.2.2			Slope area cover material: Load, haul, dump along crest	59,101	Cm3	R.051	964.8	$0.72	$42,444	$2.37	$140,125	$0.80	49,031	$47,070	$0.00	$0	$3.89	$229,639			
A1.1.2.3			Slope area cover material: Spread down slope	179	hrs	C.2.13	178.9	$44.39	$7,941	$194.68	$34,828	$67.01	12,487	$11,988	$0.00	$0	$306.07	$54,757			
A1.1.3.1		Revegetate	Seed/Fertilize: tractor application	42	ha	C.6.01	1,379.7	$1,029.50	$42,879	$401.72	$16,732	$232.94	10,107	$9,702	$1,005.72	$41,889	$2,670	$111,202			
A1.1.3.2			Tree seedling application (1,000 stems/ha)	52	ha	C.6.06	2,668.2	$1,832.33	$95,397	$25.98	$1,353	$15.00	813	$781	$469.71	$24,455	$2,343	$121,985			
A1.2	*Southwest Dump (excluding high grade waste area)*																		$907,790		
A1.2.1		Re-grade	Slopes: Re-grade remaining to 3H:1V	12	hrs	C.2.13	12.1	$44.39	$536.44	$194.68	$2,353	$67.01	844	$809.81	$0.00	$0.00	$306.07	$3,699			
A1.2.2		Cover	Load, haul, dump spread	156,905	m3	R.059	2,731.6	$0.77	$120,388.52	$2.49	$391,345	$0.82	133,341	$128,007.23	$0.00	$0.00	$4.08	$639,741			
A1.2.3.1		Revegetate	Seed/Fertilize: tractor application	43.93	ha	C.6.01	1,455.4	$1,029.50	$45,229.58	$401.72	$17,649	$232.94	10,660	$10,234.04	$1,005.72	$44,184.74	$2,669.89	$117,297			
A1.2.3.2			Tree seedling application (1,000 stems/ha)	62.76	ha	C.6.06	3,216.6	$1,832.33	$115,000.98	$25.98	$1,631	$15.00	981	$941.43	$469.71	$29,479.97	$2,343.03	$147,053			
A1.3	*Southwest Dump High-Grade Waste*																		$249,632		
A1.3.1.1		Relocate High-grade waste	Load, haul, dump, spread in pit (below long-term water elevation)	45,711	Bm3	R.069	688.0	$0.66	$30,283.89	$2.19	$100,276	$0.73	34,965	$33,566.76	$0.00	$0.00	$3.59	$164,127			
A1.3.2.1		Regrade	Flat areas: Re-grade to form tertiary drainage catchments	31	hrs	C.2.13	31.1	$44.39	$1,378.38	$194.68	$6,045	$67.01	2,168	$2,080.81	$0.00	$0.00	$306.07	$9,505			
A1.3.3.1		Cover	Flat area cover material: Load, haul, dump spread	15,583	Cm3	R.060	257.8	$0.73	$11,362.15	$2.37	$36,935	$0.78	12,585	$12,081.20	$0.00	$0.00	$3.87	$60,378			
A1.3.4.1		Revegetate	Seed/Fertilize: broadcast seeding	3.12	ha	C.6.01	103.2	$1,029.50	$3,208.44	$401.72	$1,252	$232.94	756	$725.97	$1,005.72	$3,134.33	$2,669.89	$8,321			
A1.3.4.2			Tree seedling application (1,000 stems/ha)	3.12	ha	C.6.06	159.7	$1,832.33	$5,710.47	$25.98	$81	$15.00	49	$46.75	$469.71	$1,463.85	$2,343.03	$7,302			
A1.4	*Main Pit SAT*																		$4,103,230		
A1.4.1.1		Relocate SAT below water	Load, haul, dump, spread in pit	1,040,000	Cm3	R.012	17,781.5	$0.75	$785,076.51	$1.86	$1,935,773	$0.51	555,290	$533,078.26	$0.00	$0.00	$3.13	$3,253,928			
A1.4.1.2			Allowance for additional SAT material from SAT Dump	30,000	Cm3	R.012	512.9	$0.75	$22,646.44	$1.86	$55,840	$0.51	16,018	$15,377.26	$0.00	$0.00	$3.13	$93,863			
A1.4.2.1		Backfill SAT Excavation	Load, haul, dump WR to form buttress below MPD	245,000	Cm3	R.013	3,226.9	$0.58	$142,343.39	$1.89	$462,905	$0.61	156,449	$150,191.13	$0.00	$0.00	$3.08	$755,439			Cover and revegetation of backfill included in MPD (A1.5)
A1.5	*Main Pit Dump (Incl. SWB, In-Pit Dumps)*																		$686,521		
A1.5.1.1		Regrade	Flat areas: Re-grade to form tertiary drainage catchments	65	hrs	C.2.13	65.1	$44.39	$2,891.84	$194.68	$12,683	$67.01	4,547	$4,365.54	$0.00	$0.00	$306.07	$19,941			Assumes MPD is fully constructed
A1.5.1.2			Slopes: Re-grade to 3H:1V	207	hrs	C.2.13	206.7	$44.39	$9,174.86	$194.68	$40,240	$67.01	14,428	$13,850.42	$0.00	$0.00	$306.07	$63,265			
A1.5.2.1		Cover	Flat area cover material: Load, haul, dump spread	45,270	m3	R.053	732.0	$0.71	$32,263.18	$2.32	$104,877	$0.76	35,734	$34,304.93	$0.00	$0.00	$3.79	$171,445			
A1.5.2.2			Slope area cover material: Load, haul, dump along crest	84,053	m3	R.054	1,165.0	$0.61	$51,284.32	$2.02	$169,813	$0.68	59,212	$56,843.70	$0.00	$0.00	$3.31	$277,941			
A1.5.2.3			Slope area cover material: Spread down slope	79	hrs	C.2.13	79.3	$44.39	$3,520.01	$194.68	$15,438	$67.01	5,535	$5,313.82	$0.00	$0.00	$306.07	$24,272			
A1.5.3.1		Revegetate	Seed/Fertilize: broadcast seeding	25.9	ha	C.6.01	856.8	$1,029.50	$26,627.72	$401.72	$10,390	$232.94	6,276	$6,025.02	$1,005.72	$26,012.60	$2,669.89	$69,056			
A1.5.3.2			Tree seedling application (1,000 stems/ha)	25.9	ha	C.6.06	1,325.6	$1,832.33	$47,392.66	$25.98	$672	$15.00	404	$387.97	$469.71	$12,148.89	$2,343.03	$60,602			
A1.6.0.0	*Mill Valley Fill Extension (Stage 1 and 2)*																		$124,446		
A1.6.1.1		Complete Fill Placement	Waste Rock fill: Load, haul, dump spread	0	m3	R.041	0.0	$0.97	$0.00	$3.15	$0	$1.04	0	$0.00	$0.00	$0.00	$5.15	$0			Quantitiies updated in late 2017 to reflect progressive
A1.6.2.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	81	hrs	C.2.13	81.2	$44.39	$3,605.65	$194.68	$15,814	$67.01	5,670	$5,443.12	$0.00	$0.00	$306.07	$24,863			reclamation completed at the facility.
A1.6.2.2			Slopes: Re-grade to 3H:1V	4	hrs	C.2.13	4.3	$44.39	$189.50	$194.68	$831	$67.01	298	$286.07	$0.00	$0.00	$306.07	$1,307			
A1.6.3.1		Cover	Flat area cover material: Load, haul, dump spread	5,743	m3	R.056	116.9	$0.90	$5,150.07	$2.91	$16,736	$0.96	5,737	$5,507.52	$0.00	$0.00	$4.77	$27,394			
A1.6.3.2			Slope area cover material: Load, haul, dump along crest	1,496	m3	R.057	26.7	$0.78	$1,172.86	$2.59	$3,872	$0.87	1,355	$1,300.68	$0.00	$0.00	$4.24	$6,346			
A1.6.3.3			Slope area cover material: Spread down slope	1	hrs	C.2.13	0.9	$44.39	$40.75	$194.68	$179	$67.01	64	$61.52	$0.00	$0.00	$306.07	$281			
A1.6.4.1		Revegetate	Seed/Fertilize: broadcast seeding	8.74	ha	C.6.01	289.4	$1,029.50	$8,992.98	$401.72	$3,509	$232.94	2,120	$2,034.83	$1,005.72	$8,785.24	$2,669.89	$23,322			
A1.6.4.2			Tree seedling application (1,000 stems/ha)	17.47	ha	C.6.06	895.4	$1,832.33	$32,011.85	$25.98	$454	$15.00	273	$262.06	$469.71	$8,206.09	$2,343.03	$40,934			
A2	**Overburden Dumps**																			**$370,112**	
A2.1	*Area 118 Pit Backfill Dump*																		$25,293		
A2.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	14	hrs	C.2.13	14.0	$44.39	$621.99	$194.68	$2,728	$67.01	978	$938.96	$0.00	$0.00	$306.07	$4,289			
A2.2.2.1		Revegetate	Seed/Fertilize: broadcast seeding	4	ha	C.6.01	138.8	$1,029.50	$4,313.51	$401.72	$1,683	$232.94	1,017	$976.01	$1,005.72	$4,213.87	$2,669.89	$11,187			
A2.2.2.2			Tree seedling application (1,000 stems/ha)	4	ha	C.6.06	214.7	$1,832.33	$7,677.29	$25.98	$109	$15.00	65	$62.85	$469.71	$1,968.04	$2,343.03	$9,817			
A2.2	*Ice-rich Overburden Dump*																		$13,827		
A2.2.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A2.2.1.2			Berm: Regrade to 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			WR Berm area included in Southwest Dump
A2.2.3.1		Revegetate	Seed/Fertilize: broadcast seeding	2.76	ha	C.6.01	91.4	$1,029.50	$2,839.68	$401.72	$1,108	$232.94	669	$642.53	$1,005.72	$2,774.08	$2,669.89	$7,364			
A2.2.3.2			Tree seedling application (1,000 stems/ha)	2.76	ha	C.6.06	141.4	$1,832.33	$5,054.12	$25.98	$72	$15.00	43	$41.37	$469.71	$1,295.60	$2,343.03	$6,463			
A2.3	*Reclamation Overburden Dump*																		$330,992		
A2.3.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	594	hrs	C.2.13	593.6	$44.39	$26,347.50	$194.68	$115,557	$67.01	41,432	$39,774.35	$0.00	$0.00	$306.07	$181,679			
A2.3.1.2			Slopes: Regrade to 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			Assumes borrow removed in such a manner that significant resloping is not required.
A2.3.2.1		Revegetate	Seed/Fertilize: broadcast seeding	29.79	ha	C.6.01	986.7	$1,029.50	$30,664.45	$401.72	$11,966	$232.94	7,228	$6,938.40	$1,005.72	$29,956.07	$2,669.89	$79,525			
A2.3.2.2			Tree seedling application (1,000 stems/ha)	29.79	ha	C.6.06	1,526.5	$1,832.33	$54,577.32	$25.98	$774	$15.00	465	$446.79	$469.71	$13,990.64	$2,343.03	$69,789			
A2.4	*Ridgetop South Backfill Dump*																		$0		
A3	**Ore Stockpiles**																			**$1,419,486**	
A3.1	*High and Medium Grade Sulphide Ore Stockpile (South of Mill)*																		$1,264,472		(North, West, East, and South Stockpiles)
A3.1.1.1		Relocate to Pit	Ore: Load, haul, dump, spread into Main Pit	205,490	m3	R.044	2,417.3	$0.52	$106,498.23	$1.72	$354,091	$0.57	122,871	$117,956.06	$0.00	$0.00	$2.82	$578,546			
A3.1.1.2			Ore Pads (over-excavate): Load, haul, dump spread into Main Pit	71,007	m3	R.044	835.3	$0.52	$36,800.28	$1.72	$122,356	$0.57	42,458	$40,759.51	$0.00	$0.00	$2.82	$199,915			
A3.1.1.3			Complete confirmation testing	1,421	ea	C.3.02	236.8	$7.43	$10,556.37	$1.33	$1,883	$0.00	0	$0.00	$13.26	$18,842.46	$22.01	$31,282			
A3.1.2.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	289	hrs	C.2.13	289.4	$44.39	$12,843.71	$194.68	$56,331	$67.01	20,197	$19,388.95	$0.00	$0.00	$306.07	$88,564			
A3.1.3.1		Cover	Cover material: Load, haul, dump spread (0.5m)	69,653	m3	R.055	1,264.9	$0.80	$55,711.18	$2.60	$181,043	$0.86	62,060	$59,577.97	$0.00	$0.00	$4.25	$296,332			
A3.1.4.1		Revegetate	Seed/Fertilize: broadcast seeding	13.93	ha	C.6.01	461.5	$1,029.50	$14,341.67	$401.72	$5,596	$232.94	3,380	$3,245.07	$1,005.72	$14,010.36	$2,669.89	$37,193			
A3.1.4.2			Tree seedling application (1,000 stems/ha)	13.93	ha	C.6.06	713.9	$1,832.33	$25,525.64	$25.98	$362	$15.00	218	$208.96	$469.71	$6,543.38	$2,343.03	$32,640			
A3.2	*Low Grade Sulphide Ore Stockpile (Main Dump Area)*																		$155,014		
A3.2.1.1		Relocate to Pit	Ore: Load, haul, dump, spread into Main Pit	15,490	m3	R.036	195.9	$0.56	$8,624.35	$1.84	$28,557	$0.62	9,958	$9,559.25	$0.00	$0.00	$3.02	$46,741			
A3.2.1.2			Ore Pad (over-excavate): Load, haul, dump spread into Main Pit	31,316	m3	R.036	396.1	$0.56	$17,435.36	$1.84	$57,732	$0.62	20,131	$19,325.41	$0.00	$0.00	$3.02	$94,493			
A3.2.1.3			Complete confirmation testing	626	ea	C.3.02	104.3	$7.43	$4,650.45	$1.33	$829	$0.00	0	$0.00	$13.26	$8,300.76	$22.01	$13,781			
A3.2.2.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			Included in Main Waste Dump Closure
A3.2.3.1		Cover	Cover material: Load, haul, dump spread	0	m3	R.050	0.0	$0.82	$0.00	$2.67	$0	$0.88	0	$0.00	$0.00	$0.00	$4.37	$0			Included in Main Waste Dump Closure
A3.2.4.1		Revegetate	Seed/Fertilize: broadcast seeding	0	ha	C.6.01	0.0	$1,029.50	$0.00	$401.72	$0	$232.94	0	$0.00	$1,005.72	$0.00	$2,669.89	$0			Included in Main Waste Dump Closure
A3.2.4.2			Tree seedling application (1,000 stems/ha)	0	ha	C.6.06	0.0	$1,832.33	$0.00	$25.98	$0	$15.00	0	$0.00	$469.71	$0.00	$2,343.03	$0			Included in Main Waste Dump Closure
A4	**Open Pits**																			**$177,927**	
A4.1	*Main Pit*																		$150,984		
A4.1.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	7,500	m2	C.2.03	28.5	$0.17	$1,265.55	$0.49	$3,658	$0.13	1,048	$1,005.87	$0.00	$0.00	$0.79	$5,930			
A4.1.1.2			Safety Berm: Construct around highwall perimeter	4,860	m3	C.2.02	45.8	$0.42	$2,035.14	$0.89	$4,333	$0.25	1,267	$1,215.92	$0.00	$0.00	$1.56	$7,584			
A4.1.1.3			Place large boulders across any pit access points	52	m3	R.001	1.6	$1.39	$72.20	$3.03	$157	$0.90	49	$46.76	$0.00	$0.00	$5.31	$276			
A4.1.1.4			Allowance: Install warning signs around pit perimeter at key locations	6	ea	C.5.13	5.6	$40.27	$241.63	$15.90	$95	$5.57	35	$33.39	$199.26	$1,195.57	$261.00	$1,566			
A4.1.2.1		Exposed tailings cover	Trafficability Layer: Load, haul, dump spread	17,750	m3	R.014	554.4	$1.38	$24,425.58	$2.86	$50,708	$0.82	15,091	$14,487.41	$0.00	$0.00	$5.05	$89,621			
A4.1.2.2			Overburden Layer: Load, haul, dump spread	8,875	m3	R.052	158.4	$0.79	$6,976.73	$2.55	$22,672	$0.84	7,772	$7,460.97	$0.00	$0.00	$4.18	$37,110			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A4.1.3.1		Revegetate	Seed/Fertilize: broadcast seeding	2	ha	C.6.01	58.8	$1,029.50	$1,827.37	$401.72	$713	$232.94	431	$413.48	$1,005.72	$1,785.15	$2,669.89	$4,739			
A4.1.3.2			Tree seedling application (1,000 stems/ha)	2	ha	C.6.06	91.0	$1,832.33	$3,252.39	$25.98	$46	$15.00	28	$26.63	$469.71	$833.74	$2,343.03	$4,159			
A4.2	*Area 2 Pit*																		$10,001		
A4.2.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	0	m2	C.2.03	0.0	$0.17	$0.00	$0.49	$0	$0.13	0	$0.00	$0.00	$0.00	$0.79	$0			
A4.2.1.2			Safety Berm: Construct around highwall perimeter	5,317	m3	C.2.02	50.2	$0.42	$2,226.54	$0.89	$4,740	$0.25	1,386	$1,330.27	$0.00	$0.00	$1.56	$8,297			
A4.2.1.3			Place large boulders across any pit access points	26	m3	R.001	0.8	$1.39	$36.10	$3.03	$79	$0.90	24	$23.38	$0.00	$0.00	$5.31	$138			
A4.2.1.4			Allowance: Install warning signs around pit perimeter at key locations	6	ea	C.5.13	5.6	$40.27	$241.63	$15.90	$95	$5.57	35	$33.39	$199.26	$1,195.57	$261.00	$1,566			
A4.3	*Area 118 Pit*																		$5,461		
A4.3.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	2,580	m2	C.2.03	9.8	$0.17	$435.35	$0.49	$1,258	$0.13	360	$346.02	$0.00	$0.00	$0.79	$2,040			
A4.3.1.2			Safety Berm: Construct around highwall perimeter	1,435	m3	C.2.02	13.5	$0.42	$600.85	$0.89	$1,279	$0.25	374	$358.99	$0.00	$0.00	$1.56	$2,239			
A4.3.1.3			Place large boulders across any pit access points	26	m3	R.001	0.8	$1.39	$36.10	$3.03	$79	$0.90	24	$23.38	$0.00	$0.00	$5.31	$138			
A4.3.1.4			Allowance: Install warning signs around pit perimeter at key locations	4	ea	C.5.13	3.7	$40.27	$161.09	$15.90	$64	$5.57	23	$22.26	$199.26	$797.04	$261.00	$1,044			
A4.4	*Minto North Pit*																		$11,481		
A4.4.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	5,900	m2	C.2.03	22.4	$0.17	$995.56	$0.49	$2,878	$0.13	824	$791.29	$0.00	$0.00	$0.79	$4,665			
A4.4.1.2			Safety Berm: Construct around highwall perimeter	3,610	m	C.2.02	34.1	$0.42	$1,511.81	$0.89	$3,219	$0.25	941	$903.25	$0.00	$0.00	$1.56	$5,634			
A4.4.1.3			Place large boulders across any pit access points	26	m3	R.001	0.8	$1.39	$36.10	$3.03	$79	$0.90	24	$23.38	$0.00	$0.00	$5.31	$138			
A4.4.1.4			Allowance: Install warning signs around pit perimeter at key locations	4	ea	C.5.13	3.7	$40.27	$161.09	$15.90	$64	$5.57	23	$22.26	$199.26	$797.04	$261.00	$1,044			
A5	**Underground Openings**																			**$81,637**	
A5.1	*Minto North Portal*																		$16,573		
A5.1.1.1		Seal portal	Load, Haul, Dump backfill plug	2,588	m3	R.001	81.5	$1.39	$3,592.52	$3.03	$7,832	$0.90	2,423	$2,326.51	$0.00	$0.00	$5.31	$13,751			Material sourced from nearby laydown area rockfill
A5.1.1.2			Doze backfill plug into portal with small dozer	16	hrs	C.2.19	0.0	$44.39	$723.90	$105.36	$1,718	$23.28	395	$379.66	$0.00	$0.00	$173.02	$2,822			
A5.2	*Minto North Portal Egress*																		$16,573		
A5.2.1.1		Seal portal	Load, Haul, Dump backfill plug	2,588	m3	R.001	81.5	$1.39	$3,592.52	$3.03	$7,832	$0.90	2,423	$2,326.51	$0.00	$0.00	$5.31	$13,751			Material sourced from nearby laydown area rockfill
A5.2.1.2			Doze backfill plug into portal with small dozer	16	hrs	C.2.19	0.0	$44.39	$723.90	$105.36	$1,718	$23.28	395	$379.66	$0.00	$0.00	$173.02	$2,822			
A5.3	*Minto South Portal*																		$16,573		
A5.3.1.1		Seal portal	Load, Haul, Dump backfill plug	2,588	m3	R.001	81.5	$1.39	$3,592.52	$3.03	$7,832	$0.90	2,423	$2,326.51	$0.00	$0.00	$5.31	$13,751			Material sourced from nearby laydown area rockfill
A5.3.1.2			Doze backfill plug into portal with small dozer	16	hrs	C.2.19	0.0	$44.39	$723.90	$105.36	$1,718	$23.28	395	$379.66	$0.00	$0.00	$173.02	$2,822			
A5.3.2.1		Revegetate	Included in Yards (WBS No. A11-10)																		
A5.4	*Area 118 Vent Raise*																		$10,691		
A5.4.1.1		Site Preparation	Removal of Heater/Fan included in Demolition																		
A5.4.1.2			Excavate and clean perimeter around shaft to suitable material.	4	m3	C.8.01	8.0	$84.00	$335.99	$13.66	$55	$6.83	28	$27.33	$0.00	$0.00	$104.49	$418			
A5.4.2.1		Concrete Cap installation	Structural steel - Supply and install I-Beam structural support	8	m	C.8.02	9.8	$50.19	$401.48	$15.95	$128	$6.05	50	$48.37	$16.09	$128.68	$88.27	$706			
A5.4.2.2			Formwork: Supply and install concrete ring-wall formwork	8	m2	C.8.03	51.3	$309.43	$2,475.43	$0.00	$0	$0.00	0	$0.00	$3.54	$28.32	$312.97	$2,504			
A5.4.2.3			Ringwall Rebar: supply and install	32	m	C.8.05	0.8	$1.27	$40.74	$0.00	$0	$0.00	0	$0.00	$3.37	$107.82	$4.64	$149			
A5.4.2.4			Concrete slab: supply and install	8.6	m3	C.8.06	61.1	$329.23	$2,844.57	$10.63	$92	$3.50	31	$30.21	$386.22	$3,336.96	$729.58	$6,304			Includes slab formwork, rebar, finishing.
A5.4.2.5			Vent Raise Pipe Supply and install	1	ea.	C.8.07	0.5	$17.77	$17.77	$0.00	$0	$0.00	0	$0.00	$460.62	$460.62	$478.39	$478			
A5.4.3.1		Backfill cap area	Backfill: Load, haul, dump, place over concrete cap	25	m3	R.001	0.8	$1.39	$34.71	$3.03	$76	$0.90	23	$22.48	$0.00	$0.00	$5.31	$133			
A5.5	*Minto East Vent Raise*																		$10,610		
A5.5.1.2		Preparation	Excavate and clean perimeter around shaft to suitable material.	4	m3	C.8.01	8.0	$84.00	$335.99	$13.66	$55	$6.83	28	$27.33	$0.00	$0.00	$104.49	$418			
A5.5.2.1		Concrete Cap installation	Structural steel - Supply and install I-Beam structural support	8	m	C.8.02	9.8	$50.19	$401.48	$15.95	$128	$6.05	50	$48.37	$16.09	$128.68	$88.27	$706			
A5.5.2.2			Formwork: Supply and install concrete ring-wall formwork	8	m2	C.8.03	51.3	$309.43	$2,475.43	$0.00	$0	$0.00	0	$0.00	$3.54	$28.32	$312.97	$2,504			
A5.5.2.3			Ringwall Rebar: supply and install	14	m	C.8.05	0.3	$1.27	$18.33	$0.00	$0	$0.00	0	$0.00	$3.37	$48.52	$4.64	$67			
A5.5.2.4			Concrete slab: supply and install	8.6	m3	C.8.06	61.1	$329.23	$2,844.57	$10.63	$92	$3.50	31	$30.21	$386.22	$3,336.96	$729.58	$6,304			Includes slab formwork, rebar, finishing.
A5.5.2.5			Vent Raise Pipe Supply and install	1	ea.	C.8.07	0.5	$17.77	$17.77	$0.00	$0	$0.00	0	$0.00	$460.62	$460.62	$478.39	$478			
A5.5.3.1		Backfill cap area	Backfill: Load, haul, dump, place over concrete cap	25	m3	R.001	0.8	$1.39	$34.71	$3.03	$76	$0.90	23	$22.48	$0.00	$0.00	$5.31	$133			
A5.6	*Minto East 2 Vent Raise*																		$10,617		
A5.6.1.1		Site Preparation	Removal of Heater/Fan included in Demolition																		
A5.6.1.2			Excavate and clean perimeter around shaft to suitable material.	4	m3	C.8.01	8.0	$84.00	$335.99	$13.66	$55	$6.83	28	$27.33	$0.00	$0.00	$104.49	$418			
A5.6.2.1		Concrete Cap installation	Structural steel - Supply and install I-Beam structural support	8	m	C.8.02	9.8	$50.19	$401.48	$15.95	$128	$6.05	50	$48.37	$16.09	$128.68	$88.27	$706			
A5.6.2.2			Formwork: Supply and install concrete ring-wall formwork	8	m2	C.8.03	51.3	$309.43	$2,475.43	$0.00	$0	$0.00	0	$0.00	$3.54	$28.32	$312.97	$2,504			
A5.6.2.3			Ringwall Rebar: supply and install	16	m	C.8.05	0.4	$1.27	$20.37	$0.00	$0	$0.00	0	$0.00	$3.37	$53.91	$4.64	$74			
A5.6.2.4			Concrete slab: supply and install	8.6	m3	C.8.06	61.1	$329.23	$2,844.57	$10.63	$92	$3.50	31	$30.21	$386.22	$3,336.96	$729.58	$6,304			Includes slab formwork, rebar, finishing.
A5.6.2.5			Vent Raise Pipe Supply and install	1	ea.	C.8.07	0.5	$17.77	$17.77	$0.00	$0	$0.00	0	$0.00	$460.62	$460.62	$478.39	$478			
A5.6.3.1		Backfill cap area	Backfill: Load, haul, dump, place over concrete cap	25	m3	R.001	0.8	$1.39	$34.71	$3.03	$76	$0.90	23	$22.48	$0.00	$0.00	$5.31	$133			
A6	**External Tailings Facilities**																			**$202,656**	
A6.1	*Dry Stack Tailings Storage Facility*																		$202,656		
A6.1.1.1		Waste Rock Shell	Regrade WR shell slopes to 4H:1V	3	hrs	C.2.13	2.6	$44.39	$117.60	$194.68	$516	$67.01	185	$177.52	$0.00	$0.00	$306.07	$811			
A6.1.1.2			Cover material: Load, haul, dump spread (0.5m)	5,296	m3	R.048	102.8	$0.86	$4,529.70	$2.78	$14,720	$0.91	5,046	$4,844.09	$0.00	$0.00	$4.55	$24,094			
A6.1.1.3			Scarify surface	0	m2	C.2.16	0.0	$0.03	$0.00	$0.07	$0	$0.02	0	$0.00	$0.00	$0.00	$0.12	$0			
A6.1.1.4			Seed/Fertilize: broadcast seeding	1.06	ha	C.6.01	35.1	$1,029.50	$1,090.39	$401.72	$425	$232.94	257	$246.72	$1,005.72	$1,065.20	$2,669.89	$2,828			
A6.1.1.5			Tree seedling application (1,000 stems/ha)	5.30	ha	C.6.06	271.4	$1,832.33	$9,703.52	$25.98	$138	$15.00	83	$79.44	$469.71	$2,487.45	$2,343.03	$12,408			
A6.1.2.1		Fill Area at south end of DS	Load, haul, dump, spread	9,090	Cm3	R.001	286.5	$1.39	$12,620.67	$3.03	$27,515	$0.90	8,514	$8,173.14	$0.00	$0.00	$5.31	$48,309			Fill from TDD road and/or unsuitable DSTSF cover material
A6.1.3.1		Tailings surface	Regrade tailings area south of the tailings	2	hrs	C.2.13	1.6	$44.39	$71.53	$194.68	$314	$67.01	112	$107.98	$0.00	$0.00	$306.07	$493			
A6.1.3.2			Regrade surface to form tertiary drainage catchments	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A6.1.3.3			Cover: Excavate unsuitable temporary cover material	0	m3	R.003	0.0	$0.51	$0.00	$1.70	$0	$0.57	0	$0.00	$0.00	$0.00	$2.78	$0			
A6.1.3.4			Cover material: Load, haul, dump spread (0.5m)	3,887	m3	R.048	75.5	$0.86	$3,324.91	$2.78	$10,805	$0.91	3,704	$3,555.68	$0.00	$0.00	$4.55	$17,685			
A6.1.3.5			Seed/Fertilize: broadcast seeding	19.16	ha	C.6.01	634.6	$1,029.50	$19,721.25	$401.72	$7,695	$232.94	4,648	$4,462.30	$1,005.72	$19,265.67	$2,669.89	$51,145			
A6.1.3.6			Tree seedling application (1,000 stems/ha)	19.16	ha	C.6.06	981.8	$1,832.33	$35,100.35	$25.98	$498	$15.00	299	$287.34	$469.71	$8,997.81	$2,343.03	$44,883			
A7	**Roads**																			**$136,050**	
A7.1	*Exploration Roads*																		$33,589		
A7.1.1.1		Regrade/Scarify	Scarify road surface	54,285	m2	C.2.16	0.0	$0.03	$1,886.30	$0.07	$3,564	$0.02	965	$926.07	$0.00	$0.00	$0.12	$6,377			
A7.1.2.1		Revegetate	Seed/Fertilize: broadcast seeding	5.43	ha	C.6.01	179.8	$1,029.50	$5,588.65	$401.72	$2,181	$232.94	1,317	$1,264.54	$1,005.72	$5,459.55	$2,669.89	$14,494			
A7.1.2.2			Tree seedling application (1,000 stems/ha)	5.43	ha	C.6.06	278.2	$1,832.33	$9,946.82	$25.98	$141	$15.00	85	$81.43	$469.71	$2,549.82	$2,343.03	$12,719			
A7.2	*Access Roads (excl. Main Site Access Road)*																		$9,082		
A7.2.1.1		Regrade/Scarify	Regrade side slopes	9	hrs	C.2.13	9.5	$44.39	$419.85	$194.68	$1,841	$67.01	660	$633.81	$0.00	$0.00	$306.07	$2,895			
A7.2.1.2			Scarify road surface	52,636	m2	C.2.16	0.0	$0.03	$1,829.00	$0.07	$3,456	$0.02	935	$897.94	$0.00	$0.00	$0.12	$6,183			
A7.2.2.1		Revegetate	Seed/Fertilize: broadcast seeding	0.00	ha	C.6.01	0.0	$1,029.50	$0.77	$401.72	$0	$232.94	0	$0.17	$1,005.72	$0.75	$2,669.89	$2			
A7.2.2.2			Tree seedling application (1,000 stems/ha)	0.00	ha	C.6.06	0.0	$1,832.33	$1.36	$25.98	$0	$15.00	0	$0.01	$469.71	$0.35	$2,343.03	$2			
A7.3	*Main Site Access Road*																		$1,343		
A7.3.1.1		New Access over MVFE	Regrade new access through MVFE	0	m3																
A7.3.2.1		Signage	Install large sign on east and west side of barge landing on Yukon River	2	ea.	C.5.14	1.9	$40.27	$80.54	$15.90	$32	$5.57	12	$11.13	$609.74	$1,219.48	$671.48	$1,343			
A7.4	*Haul Roads*																		$72,436		
A7.4.1.1		Regrade/Scarify	Regrade side slopes	21	hrs	C.2.13	20.5	$44.39	$910.71	$194.68	$3,994	$67.01	1,432	$1,374.82	$0.00	$0.00	$306.07	$6,280			
A7.4.1.2			Scarify road surface	91,084	m2	C.2.16	0.0	$0.03	$3,164.99	$0.07	$5,981	$0.02	1,619	$1,553.84	$0.00	$0.00	$0.12	$10,700			
A7.4.2.1		Revegetate	Seed/Fertilize: broadcast seeding	11.06	ha	C.6.01	366.5	$1,029.50	$11,389.06	$401.72	$4,444	$232.94	2,684	$2,576.99	$1,005.72	$11,125.97	$2,669.89	$29,536			
A7.4.2.2			Tree seedling application (1,000 stems/ha)	11.06	ha	C.6.06	567.0	$1,832.33	$20,270.53	$25.98	$287	$15.00	173	$165.94	$469.71	$5,196.25	$2,343.03	$25,920			
A7.5	*Culvert Removal*																		$19,600		
A7.5.1.1		Excavate culverts	Mobilization allowance for excavation fleet to various sites	6	hrs	C.7.01	12.0	$88.22	$529.34	$118.29	$710	$44.16	276	$264.96	$0.00	$0.00	$250.67	$1,504			
A7.5.1.2			Load, haul, dump, spread (spoil locally)	18,020	m3	R.901	59.1	$0.15	$2,622.79	$0.58	$10,499	$0.20	3,841	$3,687.10	$0.00	$0.00	$0.93	$16,809			
A7.5.1.2		Waste disposal	On-site disposal (demolition debris, etc.)	190	Lm3	R.037	8.1	$1.87	$356.36	$3.75	$713	$1.14	227	$217.54	$0.00	$0.00	$6.77	$1,287			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A8	**Demolition**																			**$1,976,253**	
A8.1	*Airstrip Area*																		$21,059		
A8.1.1.1		Remove tanks/equipment	Dismantle and prep for transport	7	hrs	C.1.13	28.0	$168.00	$1,175.97	$86.19	$603	$14.53	106	$101.70	$0.00	$0.00	$268.71	$1,881			
A8.1.2.1		Waste Oil Tanker Secondar	Bedding Material: Load, haul dump bedding material to landfarm	47	Cm3	R.001	1.5	$1.39	$65.73	$3.03	$143	$0.90	44	$42.57	$0.00	$0.00	$5.31	$252			
A8.1.2.2			Cut and fold liner	219	m2	C.1.15	2.5	$0.43	$94.83	$0.19	$43	$0.07	16	$15.13	$0.00	$0.00	$0.70	$152			
A8.1.2.3			Regrade area to promote positive drainage	192	m2	C.2.11	1.9	$0.44	$85.22	$1.28	$246	$0.35	71	$67.74	$0.00	$0.00	$2.08	$399			
A8.1.3.1		Prepare for demolition	Remove hazardous materials/prep for transport offsite	10	hrs	C.1.22	40.0	$159.30	$1,592.98	$95.83	$958	$26.93	281	$269.28	$0.00	$0.00	$282.05	$2,821			
A8.1.3.2			Disconnect services	4	hrs	C.1.09	12.0	$88.07	$352.28	$7.55	$30	$4.80	20	$19.20	$0.00	$0.00	$100.42	$402			
A8.1.4.1		Demolition	Structural building demolition: wooden buildings/tents	1,149	m3	C.1.08	87.7	$2.88	$3,310.74	$1.60	$1,834	$0.57	680	$652.51	$0.00	$0.00	$5.05	$5,797			
A8.1.4.2			Other demolition: covered storage, debris, etc.	530	m3	C.1.07	52.5	$3.74	$1,980.53	$2.07	$1,097	$0.74	407	$390.34	$0.00	$0.00	$6.55	$3,468			
A8.1.5.1		Waste disposal	On-site disposal (demolition debris, etc.)	379	m3	R.006	12.8	$1.48	$562.06	$2.95	$1,117	$0.92	361	$346.89	$0.00	$0.00	$5.35	$2,026			
A8.1.5.2			Off-site disposal (re-usable equipment, etc.)	4	m3	C.7.07	34.6	$355.10	$1,536.93	$423.68	$1,834	$113.33	511	$490.50	$0.00	$0.00	$892.11	$3,861			
A8.2	*Airport Laydown Area*																		$33,408		
A8.2.1.1		Remove tanks/equipment	Dismantle and prep for transport	2	hrs	C.1.13	8.0	$168.00	$335.99	$86.19	$172	$14.53	30	$29.06	$0.00	$0.00	$268.71	$537			
A8.2.2.1		Prepare for demolition	Remove hazardous materials/prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.2.2.2			Disconnect services	4	hrs	C.1.09	12.0	$88.07	$352.28	$7.55	$30	$4.80	20	$19.20	$0.00	$0.00	$100.42	$402			
A8.2.3.1		Demolition	Structural building demolition: steel structures	96	tonnes	C.1.05	191.3	$69.36	$6,633.00	$139.08	$13,301	$33.84	3,371	$3,236.25	$0.00	$0.00	$242.28	$23,170			
A8.2.3.2			Other demolition: covered storage etc.	560	m3	C.1.07	55.5	$3.74	$2,093.92	$2.07	$1,160	$0.74	430	$412.69	$0.00	$0.00	$6.55	$3,667			
A8.2.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	393	Lm3	R.005	15.8	$1.76	$693.24	$3.53	$1,387	$1.08	441	$423.19	$0.00	$0.00	$6.37	$2,504			
A8.2.4.2			Off-site disposal (re-usable equipment, etc.)	3	trips	C.7.07	23.0	$355.10	$1,020.92	$423.68	$1,218	$113.33	339	$325.82	$0.00	$0.00	$892.11	$2,565			
A8.3	*Camp Area*																		$375,774		
A8.3.1.1		Equipment Removal	Remove salvageable equipment	48	hrs	C.1.13	192.0	$168.00	$8,063.76	$86.19	$4,137	$14.53	726	$697.34	$0.00	$0.00	$268.71	$12,898			
A8.3.2.1		Remove modular buildings	Dismantle and prep for transport	92	ea.	C.1.20	1,794.0	$850.98	$78,290.06	$730.60	$67,215	$166.46	15,953	$15,314.69	$0.00	$0.00	$1,748.04	$160,820			
A8.3.2.2			Transport structures off-site (Whitehorse)	92	ea.	C.7.04	736.0	$351.24	$32,314.23	$303.64	$27,935	$113.33	10,861	$10,426.18	$0.00	$0.00	$768.21	$70,675			
A8.3.3.1		Prepare for demolition	Remove hazardous materials/prep for transport offsite	1	hrs	C.1.22	4.0	$159.30	$159.30	$95.83	$96	$26.93	28	$26.93	$0.00	$0.00	$282.05	$282			
A8.3.3.2			Disconnect services	8	hrs	C.1.09	24.0	$88.07	$704.56	$7.55	$60	$4.80	40	$38.40	$0.00	$0.00	$100.42	$803			
A8.3.4.1		Demolition	Structural building demolition: Steel	175	tonnes	C.1.05	350.7	$69.36	$12,160.50	$139.08	$24,385	$33.84	6,180	$5,933.13	$0.00	$0.00	$242.28	$42,479			
A8.3.4.2			Structural building demolition: Wood/misc. structures	10,959	m3	C.1.08	836.7	$2.88	$31,584.20	$1.60	$17,497	$0.57	6,484	$6,224.86	$0.00	$0.00	$5.05	$55,306			
A8.3.4.4			Other demolition: Utilidors, etc.	466	m3	C.1.07	46.1	$3.74	$1,742.09	$2.07	$965	$0.74	358	$343.34	$0.00	$0.00	$6.55	$3,051			
A8.3.5.1		Waste disposal	On-site disposal (demolition debris, etc.)	3,867	m3	R.007	175.3	$1.99	$7,708.84	$3.99	$15,425	$1.22	4,902	$4,705.86	$0.00	$0.00	$7.20	$27,839			
A8.3.5.2			Off-site disposal (re-usable equipment, etc.)	1	trips	C.7.07	8.0	$355.10	$355.10	$423.68	$424	$113.33	118	$113.33	$0.00	$0.00	$892.11	$892			
A8.3.6.1		Demolish foundations	Break in place concrete foundations	51	m3	C.1.01	4.1	$2.39	$122.15	$9.37	$479	$2.48	132	$126.81	$0.00	$0.00	$14.24	$728			
A8.4	*Explosives Plant and Storage Areas*																		$30,611		
A8.4.1.1		Remove equipment	Small equipment: dismantle and prep for transport	37	hrs	C.1.13	149.3	$168.00	$6,271.81	$86.19	$3,218	$14.53	565	$542.38	$0.00	$0.00	$268.71	$10,032			
A8.4.2.1		Remove modular buildings	Dismantle and prep for transport	1	ea.	C.1.20	19.5	$850.98	$850.98	$730.60	$731	$166.46	173	$166.46	$0.00	$0.00	$1,748.04	$1,748			
A8.4.2.2			Transport structures off-site (Whitehorse)	1	ea.	C.7.04	8.0	$351.24	$351.24	$303.64	$304	$113.33	118	$113.33	$0.00	$0.00	$768.21	$768			
A8.4.3.1		Prepare for demolition	Remove hazardous materials/prep for transport offsite	4	hrs	C.1.22	16.0	$159.30	$637.19	$95.83	$383	$26.93	112	$107.71	$0.00	$0.00	$282.05	$1,128			
A8.4.3.2			Disconnect services	2	hrs	C.1.09	6.0	$88.07	$176.14	$7.55	$15	$4.80	10	$9.60	$0.00	$0.00	$100.42	$201			
A8.4.4.1		Demolition	Structural building demolition: Steel	47	tonnes	C.1.05	94.2	$69.36	$3,266.25	$139.08	$6,550	$33.84	1,660	$1,593.61	$0.00	$0.00	$242.28	$11,410			
A8.4.5.1		Waste disposal	On-site disposal (demolition debris, etc.)	103	m3	R.008	5.4	$2.32	$239.23	$4.65	$479	$1.42	152	$146.04	$0.00	$0.00	$8.39	$864			
A8.4.5.2			Off-site disposal (re-usable equipment, etc.)	5	trips	C.7.07	40.0	$355.10	$1,775.51	$423.68	$2,118	$113.33	590	$566.64	$0.00	$0.00	$892.11	$4,461			
A8.5	*Fuel Storage Area*																		$114,842		
A8.5.1.1		Remove equipment/dismant	Small equipment: dismantle and prep for transport	24	hrs	C.1.13	96.0	$168.00	$4,031.88	$86.19	$2,069	$14.53	363	$348.67	$0.00	$0.00	$268.71	$6,449			
A8.5.1.2			Large equipment (crane req'd): dismantle and prep for transport	192	hrs	C.1.14	1,153.4	$269.89	$51,883.26	$167.56	$32,213	$20.53	4,111	$3,946.31	$0.00	$0.00	$457.98	$88,042			
A8.5.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.5.2.2			Disconnect services	8	hrs	C.1.09	24.0	$88.07	$704.56	$7.55	$60	$4.80	40	$38.40	$0.00	$0.00	$100.42	$803			
A8.5.2.3			Clean out tanks, pressure wash, remove sludge	10	hrs	C.1.04	10.3	$35.54	$365.13	$15.75	$162	$2.00	21	$20.55	$0.00	$0.00	$53.29	$548			
A8.5.3.1		Demolish structures	Misc. debris/scraps (tanks included in dismantling above)	75	m3	C.1.08	5.7	$2.88	$216.15	$1.60	$120	$0.57	44	$42.60	$0.00	$0.00	$5.05	$378			
A8.5.4.1		Remove secondary containr	Bedding Material: Load, haul dump bedding material to landfarm	623	Cm3	R.009	32.8	$2.32	$1,442.73	$4.64	$2,887	$1.41	917	$880.71	$0.00	$0.00	$8.37	$5,210			
A8.5.4.2			Cut and fold liner	1,747	m2	C.1.15	20.3	$0.43	$758.23	$0.19	$340	$0.07	126	$120.97	$0.00	$0.00	$0.70	$1,219			
A8.5.4.3			Regrade area to promote positive drainage	1,666	m2	C.2.11	16.7	$0.44	$739.50	$1.28	$2,138	$0.35	612	$587.76	$0.00	$0.00	$2.08	$3,465			
A8.5.5.1		Waste disposal	On-site disposal (demolition debris, etc.)	229	m3	R.009	12.1	$2.32	$531.18	$4.64	$1,063	$1.41	338	$324.26	$0.00	$0.00	$8.37	$1,918			
A8.5.5.2			Off-site disposal (re-usable equipment, etc.)	7	trips	C.7.07	56.0	$355.10	$2,485.71	$423.68	$2,966	$113.33	826	$793.30	$0.00	$0.00	$892.11	$6,245			
A8.6	*Mill Area*																		$1,265,173		Excludes WTP
A8.6.1.1		Remove equipment	Small equipment: dismantle and prep for transport	379	hrs	C.1.13	1,515.0	$168.00	$63,626.87	$86.19	$32,644	$14.53	5,732	$5,502.37	$0.00	$0.00	$268.71	$101,773			
A8.6.1.2			Large equipment (crane req'd): dismantle and prep for transport	466	hrs	C.1.14	2,796.5	$269.89	$125,789.65	$167.56	$78,099	$20.53	9,966	$9,567.74	$0.00	$0.00	$457.98	$213,456			
A8.6.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	65	hrs	C.1.22	260.0	$159.30	$10,354.37	$95.83	$6,229	$26.93	1,823	$1,750.32	$0.00	$0.00	$282.05	$18,333			Off-site disposal included in 'Waste Disposal'
A8.6.2.6			Reagents: Disposal and tipping fees	1.0	ls	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A8.6.2.7			Decontaminate buildings: wash equipment/structures, etc.	38	hrs	C.1.04	37.8	$35.54	$1,342.93	$15.75	$595	$2.00	79	$75.58	$0.00	$0.00	$53.29	$2,014			
A8.6.2.8			Decontaminate buildings: Wash floors etc.	4,732	m2	C.1.03	40.8	$0.31	$1,448.16	$0.14	$642	$0.02	85	$81.50	$0.00	$0.00	$0.46	$2,171			
A8.6.3.1		Demolition	Structural building demolition: Steel	2,660	tonnes	C.1.05	5,319.1	$69.36	$184,463.18	$139.08	$369,899	$33.84	93,750	$89,999.96	$0.00	$0.00	$242.28	$644,363			
A8.6.3.2			Structural building demolition: Wood/misc. structures	960	m3	C.1.08	73.3	$2.88	$2,766.75	$1.60	$1,533	$0.57	568	$545.29	$0.00	$0.00	$5.05	$4,845			
A8.6.3.3			Other demolition: Steel	526	tonnes	C.1.05	1,052.7	$69.36	$36,505.48	$139.08	$73,204	$33.84	18,553	$17,811.10	$0.00	$0.00	$242.28	$127,520			
A8.6.3.4			Other demolition: concrete	602	m3	C.1.06	323.7	$15.92	$9,579.30	$62.48	$37,592	$16.53	10,359	$9,944.51	$0.00	$0.00	$94.92	$57,116			
A8.6.3.5			Other demolition: miscellaneous	244	m3	C.1.07	24.1	$3.74	$910.78	$2.07	$505	$0.74	187	$179.50	$0.00	$0.00	$6.55	$1,595			
A8.6.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	5,985	m3	R.037	255.1	$1.87	$11,217.23	$3.75	$22,445	$1.14	7,133	$6,847.56	$0.00	$0.00	$6.77	$40,510			
A8.6.4.2			Off-site disposal (re-usable equipment, etc.)	33	trips	C.7.07	266.0	$355.10	$11,806.23	$423.68	$14,086	$113.33	3,925	$3,767.87	$0.00	$0.00	$892	$29,660			
A8.6.5.1		Demolish foundations	Break in place concrete foundations	1,532	m3	C.1.01	123.7	$2.39	$3,659.05	$9.37	$14,359	$2.48	3,957	$3,798.55	$0.00	$0.00	$14.24	$21,817			
A8.7	*Mill Valley Fill Extension (Stage 1 and 2)*																		$28,525		
A8.7.1.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	8	hrs	C.1.22	30.0	$159.30	$1,194.73	$95.83	$719	$26.93	210	$201.96	$0.00	$0.00	$282.05	$2,115			
A8.7.1.2			Reagents: Disposal and tipping fees	1	ls	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A8.7.2.1		Demolish buildings	Structural building demolition: Wood/misc. structures (Reagent tent)	1,800	m3	C.1.08	137.4	$2.88	$5,187.66	$1.60	$2,874	$0.57	1,065	$1,022.42	$0.00	$0.00	$5.05	$9,084			
A8.7.3.1		Haul waste to landfill	Demolition debris	594	m3	R.037	25.3	$1.87	$1,113.25	$3.75	$2,228	$1.14	708	$679.58	$0.00	$0.00	$6.77	$4,020			
A8.7.3.2			Remove tires off-site for disposal	10	trips	C.7.07	78.8	$355.10	$3,495.53	$423.68	$4,171	$113.33	1,162	$1,115.57	$0.00	$0.00	$892.11	$8,782			
A8.7.3.3			Misc. debris/scraps	652	m3	R.037	27.8	$1.87	$1,221.95	$3.75	$2,445	$1.14	777	$745.94	$0.00	$0.00	$6.77	$4,413			
A8.7.4.1		Demolish foundations	Break and bury concrete foundations	8	m3	C.1.01	0.6	$2.39	$18.48	$9.37	$73	$2.48	20	$19.19	$0.00	$0.00	$14.24	$110			
A8.8	*Minto South Portal*																		$8,516		
A8.8.1.1		Remove equipment	Small equipment: dismantle and prep for transport	6	hrs	C.1.13	24.0	$168.00	$1,007.97	$86.19	$517	$14.53	91	$87.17	$0.00	$0.00	$268.71	$1,612			
A8.8.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.8.3.1		Demolition	Other demolition: miscellaneous	462	m3	C.1.07	45.7	$3.74	$1,727.03	$2.07	$957	$0.74	355	$340.38	$0.00	$0.00	$6.55	$3,024			
A8.8.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	462	m3	R.043	20.9	$1.99	$918.18	$3.98	$1,837	$1.21	584	$560.51	$0.00	$0.00	$7.18	$3,316			
A8.9	*Minto North Portal*																		$8,981		Qntys assumed to be same as Minto South Portal
A8.9.1.1		Remove equipment	Small equipment: dismantle and prep for transport	6	hrs	C.1.13	24.0	$168.00	$1,007.97	$86.19	$517	$14.53	91	$87.17	$0.00	$0.00	$268.71	$1,612			
A8.9.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.9.3.1		Demolition	Other demolition: miscellaneous	462	m3	C.1.07	45.7	$3.74	$1,727.03	$2.07	$957	$0.74	355	$340.38	$0.00	$0.00	$6.55	$3,024			
A8.9.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	462	m3	R.042	23.8	$2.27	$1,046.71	$4.53	$2,094	$1.38	666	$638.96	$0.00	$0.00	$8.18	$3,780			
A8.10	*Pelly Laydown (Includes propane tanks by W15 Sump)*																		$83,981		
A8.10.1.1		Remove modular buildings/t	Dismantle and prep for transport	7	ea.	C.1.20	136.5	$850.98	$5,956.85	$730.60	$5,114	$166.46	1,214	$1,165.25	$0.00	$0.00	$1,748.04	$12,236			
A8.10.1.2			Transport structures off-site (Whitehorse)	7	ea.	C.7.04	56.0	$351.24	$2,458.69	$303.64	$2,125	$113.33	826	$793.30	$0.00	$0.00	$768.21	$5,377			
A8.10.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.10.3.1		Demolish buildings	Structural building demolition: Steel	80	tonnes	C.1.05	159.4	$69.36	$5,527.50	$139.08	$11,084	$33.84	2,809	$2,696.88	$0.00	$0.00	$242.28	$19,309			
A8.10.3.2			Structural building demolition: Wood/misc. structures	5,672	m3	C.1.08	433.1	$2.88	$16,347.99	$1.60	$9,057	$0.57	3,356	$3,221.99	$0.00	$0.00	$5.05	$28,627			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A8.10.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	2,165	m3	R.045	100.0	$2.03	$4,395.81	$4.06	$8,796	$1.24	2,795	$2,683.42	$0.00	$0.00	$7.33	$15,875			
A8.10.4.2			Off-site disposal (re-usable equipment, etc.)	2	trips	C.7.07	15.0	$355.10	$665.82	$423.68	$794	$113.33	221	$212.49	$0.00	$0.00	$892.11	$1,673			
A8.10.5.1		Demolish foundations	Break and bury concrete foundations	23	m3	C.1.01	1.8	$2.39	$53.73	$9.37	$211	$2.48	58	$55.78	$0.00	$0.00	$14.24	$320			
A8.11	*Vent Raise - Area 118*																		$5,381		
A8.11.1.1		Remove equipment	Small equipment: dismantle and prep for transport	20	hrs	C.1.13	80.0	$168.00	$3,359.90	$86.19	$1,724	$14.53	303	$290.56	$0.00	$0.00	$268.71	$5,374			
A8.11.2.1		Waste disposal	Off-site disposal (re-usable equipment, etc.)	1	trips	R.043	0.0	$1.99	$1.99	$3.98	$4	$1.21	1	$1.21	$0.00	$0.00	$7.18	$7			
A9	**Surface Infrastructure**																			**$169,527**	
A9.1	*Tailings and Water Conveyance Pipelines*																		$91,886		
A9.1.1.1		Dismantle piping systems	Flush and clean tailings pipeline systems	8	ls	C.1.21	32.0	$203.78	$1,630.27	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$203.78	$1,630			
A9.1.1.2			Cut pipelines and prep for transport (6 to 8" HDPE pipes)	6,800	m	C.1.17	1,070.9	$7.21	$49,056.06	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$7.21	$49,056			
A9.1.1.3			Cut pipelines and prep for transport (10 to 18" HDPE pipes)	1,740	m	C.1.18	456.7	$12.02	$20,920.96	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$12.02	$20,921			
A9.1.1.4			Cut pipelines and prep for transport (20 to 36" HDPE pipes)	885	m	C.1.19	348.4	$18.04	$15,961.25	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$18.04	$15,961			
A9.2.2.1		Waste disposal	Haul pipes lines and debris to landfill	638	m3	R.037	27.2	$1.87	$1,195.62	$3.75	$2,392	$1.14	760	$729.86	$0.00	$0.00	$6.77	$4,318			
A9.2	*Conveyance Pipeline Equipment*																		$12,462		
A9.2.1.1		Dismantle piping systems	Remove pumps and prep for transport	15	ea.	C.1.24	198.0	$602.98	$9,044.72	$24.94	$374	$14.40	225	$216.00	$0.00	$0.00	$642.33	$9,635			
A9.2.1.2			Remove barges/Support Equipment prep for transport	2	ea.	C.1.23	52.8	$1,205.96	$2,411.93	$49.89	$100	$28.80	60	$57.60	$0.00	$0.00	$1,284.65	$2,569			
A9.2.2.1		Waste disposal	Transport pumps, barge and support equip. offsite (Whitehorse)	6.7	tonnes	C.7.02	2.7	$17.56	$117.67	$15.18	$102	$5.67	40	$37.96	$0.00	$0.00	$38.41	$257			
A9.3	*Powerlines*																		$65,179		
A9.3.1.1		Dismantle	Dismantle and collect powerlines	3.2	km	C.1.10	217.9	$2,639.96	$8,342.27	$940.09	$2,971	$251.90	829	$795.99	$0.00	$0.00	$3,831.94	$12,109			
A9.3.1.2			Remove power poles and load for transport off-site	60	ea.	C.1.11	548.6	$423.67	$25,420.14	$88.51	$5,311	$18.29	1,143	$1,097.14	$0.00	$0.00	$530.46	$31,828			
A9.3.1.3			Disconnect transformers and load for transport	4	ea.	C.1.12	93.3	$892.89	$3,571.55	$317.96	$1,272	$85.20	355	$340.78	$0.00	$0.00	$1,296.04	$5,184			
A9.3.2.1		Waste disposal	Transport powerlines, poles and transformers off-site	18	trips	C.7.07	144.0	$355.10	$6,391.83	$423.68	$7,626	$113.33	2,125	$2,039.90	$0.00	$0.00	$892.11	$16,058			
A10	**Water Detention Structures**																			**$786,952**	
A10.1	*W15 Sump*																		$1,049		
A10.1.1.1		Remove secondary containr	Cut and fold liner	1,310	m2	C.1.15	15.2	$0.43	$568.50	$0.19	$255	$0.07	94	$90.70	$0.00	$0.00	$0.70	$914			
A10.1.1.2			Haul liner to landfill	20	Lm3	R.070	0.8	$1.90	$37.35	$3.80	$75	$1.16	24	$22.80	$0.00	$0.00	$6.86	$135			
A10.2	*Sewage Lagoon (near IROD)*																		$23,852		
A10.2.1		Remove pond	Backfill pond	4,488	m3	R.001	141.4	$1.39	$6,231.19	$3.03	$13,585	$0.90	4,203	$4,035.32	$0.00	$0.00	$5.31	$23,852			
A10.3	*Water Storage Pond Dam*																		$762,051		
A10.3.1.1		Water Management	Install pump and pump around system	4	hrs	C.4.14	16.0	$120.43	$481.73	$20.53	$82	$6.80	28	$27.20	$0.00	$0.00	$147.76	$591			
A10.3.1.2			Pump pond water to discharge	400,000	m3	C.4.12	146.8	$0.01	$5,216.05	$0.05	$19,052	$0.01	3,058	$2,935.56	$0.00	$0.00	$0.07	$27,203			
A10.3.1.3			Maintain pump around system during WSP closure activities	103	days	C.4.13	10.3	$3.55	$364.31	$12.98	$1,331	$2.00	214	$205.03	$0.00	$0.00	$18.53	$1,900			Duration calculated as sum of associated tasks plus 20% contingency
A10.3.2.1		Breach dam	Granular zones/gen. fill: Load, haul, dump, spread in upstream North ab	62,301	m3	R.001	1,963.4	$1.39	$86,500.02	$3.03	$188,587	$0.90	58,351	$56,017.38	$0.00	$0.00	$5.31	$331,104			
A10.3.2.2			Dam Core: Load, haul, dump in temporary stockpile	23,617	m3	R.001	744.3	$1.39	$32,789.73	$3.03	$71,488	$0.90	22,119	$21,234.62	$0.00	$0.00	$5.31	$125,512			
A10.3.2.3			Rip-rap: Sort and stockpile durable rip-rap for reuse	19,285	m3	R.001	607.8	$1.39	$26,776.14	$3.03	$58,377	$0.90	18,063	$17,340.22	$0.00	$0.00	$5.31	$102,494			
A10.3.2.4			Rip-rap: Load haul, dump unsuitable rip-rap in upstream North abutmen	19,285	m3	R.001	607.8	$1.39	$26,776.14	$3.03	$58,377	$0.90	18,063	$17,340.22	$0.00	$0.00	$5.31	$102,494			
A10.3.3.1		Channel restoration	Excavate stream channel	5,584	m3	R.902	97.1	$0.77	$4,308.58	$1.64	$9,173	$0.46	2,681	$2,574.22	$0.00	$0.00	$2.88	$16,056			
A10.3.3.2			Bedding layer: Load haul, dump, place granular bedding layer along cha	726	m3	R.001	22.9	$1.39	$1,008.60	$3.03	$2,199	$0.90	680	$653.17	$0.00	$0.00	$5.31	$3,861			
A10.3.3.3			Rip-rap: Load, haul dump from temporary stockpile	964	m3	R.001	30.4	$1.39	$1,337.77	$3.03	$2,917	$0.90	902	$866.34	$0.00	$0.00	$5.31	$5,121			
A10.3.3.4			Rip-rap: place and secure	964	m3	R.904	10.5	$0.48	$465.10	$0.91	$877	$0.32	325	$311.83	$0.00	$0.00	$1.72	$1,653			
A10.3.4.1		Cover	Cover material: Load, haul, dump spread (1m)	6,975	m3	R.001	219.8	$1.39	$9,684.17	$3.03	$21,113	$0.90	6,533	$6,271.47	$0.00	$0.00	$5.31	$37,069			Cover sourced from dam core material
A10.3.5.1		Revegetate	Seed/Fertilize: broadcast seeding	1.4	ha	C.6.01	46.2	$1,029.50	$1,436.16	$401.72	$560	$232.94	338	$324.96	$1,005.72	$1,402.98	$2,669.89	$3,724			
A10.3.5.2			Tree seedling application (1,000 stems/ha)	1.4	ha	C.6.06	71.5	$1,832.33	$2,556.10	$25.98	$36	$15.00	22	$20.93	$469.71	$655.25	$2,343.03	$3,269			
A11	**Yards/Laydown Areas**																			**$758,907**	
A11.1	*Airstrip Area*																		$39,834		
A11.1.1.1		Re-grade	Re-grade slopes to be 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A11.1.1.2			Scarify surface	64,378	m2	C.2.16	0.0	$0.03	$2,236.99	$0.07	$4,227	$0.02	1,144	$1,098.24	$0.00	$0.00	$0.12	$7,562			
A11.1.2.1		Revegetate	Seed/Fertilize: broadcast seeding	6.4	ha	C.6.01	213.3	$1,029.50	$6,627.68	$401.72	$2,586	$232.94	1,562	$1,499.64	$1,005.72	$6,474.57	$2,669.89	$17,188			
A11.1.2.2			Tree seedling application (1,000 stems/ha)	6.4	ha	C.6.06	329.9	$1,832.33	$11,796.10	$25.98	$167	$15.00	101	$96.57	$469.71	$3,023.88	$2,343.03	$15,084			
A11.2	*Airport Laydown Area*																		$13,165		
A11.2.1.1		Re-grade	Scarify surface	21,277	m2	C.2.16	0.0	$0.03	$739.33	$0.07	$1,397	$0.02	378	$362.97	$0.00	$0.00	$0.12	$2,499			
A11.2.2.1		Revegetate	Seed/Fertilize: broadcast seeding	2.1	ha	C.6.01	70.5	$1,029.50	$2,190.47	$401.72	$855	$232.94	516	$495.64	$1,005.72	$2,139.87	$2,669.89	$5,681			
A11.2.2.2			Tree seedling application (1,000 stems/ha)	2.1	ha	C.6.06	109.0	$1,832.33	$3,898.65	$25.98	$55	$15.00	33	$31.92	$469.71	$999.40	$2,343.03	$4,985			
A11.3	*Camp Area*																		$118,708		
A11.3.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	7	hrs	C.2.13	6.9	$44.39	$306.20	$194.68	$1,343	$67.01	481	$462.24	$0.00	$0.00	$306.07	$2,111			
A11.3.1.2			Re-grade slopes to be 3H:1V or flatter	24	hrs	C.2.13	24.5	$44.39	$1,086.58	$194.68	$4,766	$67.01	1,709	$1,640.31	$0.00	$0.00	$306.07	$7,493			Slope between camp and main access road
A11.3.1.3			Scarify surfaces	20,626	m2	C.2.16	0.0	$0.03	$716.72	$0.07	$1,354	$0.02	367	$351.87	$0.00	$0.00	$0.12	$2,423			
A11.3.2.1		Cover	Flat area cover material: Load, haul, dump spread	10,313	Cm3	R.046	198.2	$0.85	$8,729.48	$2.75	$28,368	$0.91	9,724	$9,335.37	$0.00	$0.00	$4.50	$46,433			
A11.3.2.2			Slope area cover material: Load, haul, dump along crest	9,349	Cm3	R.047	157.2	$0.74	$6,916.70	$2.44	$22,835	$0.82	7,990	$7,670.53	$0.00	$0.00	$4.00	$37,422			
A11.3.2.3			Slope area cover material: Spread down slope	10	hrs	C.2.13	10.2	$44.39	$451.46	$194.68	$1,980	$67.01	710	$681.52	$0.00	$0.00	$306.07	$3,113			
A11.3.3.1		Revegetate	Seed/Fertilize: broadcast seeding	3.93	ha	C.6.01	130.3	$1,029.50	$4,048.49	$401.72	$1,580	$232.94	954	$916.05	$1,005.72	$3,954.97	$2,669.89	$10,499			
A11.3.3.2			Tree seedling application (1,000 stems/ha)	3.93	ha	C.6.06	201.5	$1,832.33	$7,205.60	$25.98	$102	$15.00	61	$58.99	$469.71	$1,847.12	$2,343.03	$9,214			
A11.4	*Crusher Area*																		$73,635		
A11.2.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	9	hrs	C.2.13	9.0	$44.39	$400.28	$194.68	$1,756	$67.01	629	$604.27	$0.00	$0.00	$306.07	$2,760			
A11.2.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	13,482	m3	R.055	244.8	$0.80	$10,783.37	$2.60	$35,042	$0.86	12,012	$11,531.82	$0.00	$0.00	$4.25	$57,358			
A11.2.3.1		Revegetate	Seed/Fertilize: broadcast seeding	2.70	ha	C.6.01	89.3	$1,029.50	$2,775.95	$401.72	$1,083	$232.94	654	$628.11	$1,005.72	$2,711.82	$2,669.89	$7,199			
A11.2.3.2			Tree seedling application (1,000 stems/ha)	3	ha	C.6.06	138.2	$1,832.33	$4,940.70	$25.98	$70	$15.00	42	$40.45	$469.71	$1,266.53	$2,343.03	$6,318			
A11.5	*Exploration Disturbances*																		$21,519		
A11.5.1.1		Ridgetop Area	Scarify surfaces	34,778	m2	C.2.16	0.0	$0.03	$1,208.47	$0.07	$2,284	$0.02	618	$593.29	$0.00	$0.00	$0.12	$4,085			
A11.5.1.2			Seed/Fertilize: broadcast seeding	3	ha	C.6.01	115.2	$1,029.50	$3,580.40	$401.72	$1,397	$232.94	844	$810.13	$1,005.72	$3,497.69	$2,669.89	$9,285			
A11.5.1.3			Tree seedling application (1,000 stems/ha)	3	ha	C.6.06	178.2	$1,832.33	$6,372.49	$25.98	$90	$15.00	54	$52.17	$469.71	$1,633.56	$2,343.03	$8,149			
A11.6	*Explosives Plant and Storage Areas*																		$43,165		
A11.6.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	30	hrs	C.2.13	29.6	$44.39	$1,314.39	$194.68	$5,765	$67.01	2,067	$1,984.22	$0.00	$0.00	$306.07	$9,063			
A11.6.1.2			Re-grade slopes to be 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A11.6.1.3			Scarify surfaces	55,113	m2	C.2.16	0.0	$0.03	$1,915.08	$0.07	$3,619	$0.02	979	$940.20	$0.00	$0.00	$0.12	$6,474			
A11.6.2.1		Revegetate	Seed/Fertilize: broadcast seeding	5.5	ha	C.6.01	182.6	$1,029.50	$5,673.94	$401.72	$2,214	$232.94	1,337	$1,283.83	$1,005.72	$5,542.86	$2,669.89	$14,715			
A11.6.2.2			Tree seedling application (1,000 stems/ha)	5.5	ha	C.6.06	282.5	$1,832.33	$10,098.61	$25.98	$143	$15.00	86	$82.67	$469.71	$2,588.73	$2,343.03	$12,913			
A11.7	*Fuel Storage Areas*																		$24,758		
A11.7.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	5	hrs	C.2.13	4.5	$44.39	$200.33	$194.68	$879	$67.01	315	$302.42	$0.00	$0.00	$306.07	$1,381			
A11.7.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	4,200	m3	R.049	81.8	$0.86	$3,603.19	$2.79	$11,709	$0.92	4,014	$3,853.28	$0.00	$0.00	$4.56	$19,166			
A11.7.3.1		Revegetate	Seed/Fertilize: broadcast seeding	0.84	ha	C.6.01	27.8	$1,029.50	$864.78	$401.72	$337	$232.94	204	$195.67	$1,005.72	$844.80	$2,669.89	$2,243			
A11.7.3.2			Tree seedling application (1,000 stems/ha)	1	ha	C.6.06	43.1	$1,832.33	$1,539.16	$25.98	$22	$15.00	13	$12.60	$469.71	$394.56	$2,343.03	$1,968			
A11.8	*IROD Laydown Area*																		$18,300		
A11.8.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	15	hrs	C.2.13	14.8	$44.39	$654.96	$194.68	$2,873	$67.01	1,030	$988.72	$0.00	$0.00	$306.07	$4,516			
A11.8.1.2			Re-grade slopes to be 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A11.8.1.3			Scarify surface	18,106	m2	C.2.16	0.0	$0.03	$629.15	$0.07	$1,189	$0.02	322	$308.88	$0.00	$0.00	$0.12	$2,127			
A11.8.2.1		Revegetate	Seed/Fertilize: broadcast seeding	2.33	ha	C.6.01	77.0	$1,029.50	$2,393.91	$401.72	$934	$232.94	564	$541.67	$1,005.72	$2,338.60	$2,669.89	$6,208			
A11.8.2.2			Tree seedling application (1,000 stems/ha)	2.33	ha	C.6.06	119.2	$1,832.33	$4,260.73	$25.98	$60	$15.00	36	$34.88	$469.71	$1,092.22	$2,343.03	$5,448			
A11.9	*Mill Area*																		$134,673		
A11.9.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	60	hrs	C.2.13	59.9	$44.39	$2,659.96	$194.68	$11,666	$67.01	4,183	$4,015.50	$0.00	$0.00	$306.07	$18,342			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A11.9.1.2			Scarify surface	42,365	m2	C.2.16	0.0	$0.03	$1,472.09	$0.07	$2,782	$0.02	753	$722.72	$0.00	$0.00	$0.12	$4,977			
A11.9.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	21,182	m3	R.055	384.7	$0.80	$16,942.37	$2.60	$55,057	$0.86	18,873	$18,118.30	$0.00	$0.00	$4.25	$90,118			
A11.9.3.1		Revegetate	Seed/Fertilize: broadcast seeding	4.24	ha	C.6.01	140.3	$1,029.50	$4,361.46	$401.72	$1,702	$232.94	1,028	$986.86	$1,005.72	$4,260.70	$2,669.89	$11,311			
A11.9.3.2			Tree seedling application (1,000 stems/ha)	4.24	ha	C.6.06	217.1	$1,832.33	$7,762.62	$25.98	$110	$15.00	66	$63.55	$469.71	$1,989.91	$2,343.03	$9,926			
A11.10	*Mill Valley Fill Extension (Stage 1 and 2)*																		$0		
A11.10.1		Costs included in Waste Rock Dump Section (WBS#: A1.5)		0	hrs																
A11.11	*Minto South Portal*																		$87,165		
A11.11.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	51	hrs	C.2.13	51.5	$44.39	$2,284.70	$194.68	$10,020	$67.01	3,593	$3,449.00	$0.00	$0.00	$306.07	$15,754			
A11.11.1.2			Re-grade slopes to be 3H:1V or flatter	28	hrs	C.2.13	28.0	$44.39	$1,244.58	$194.68	$5,459	$67.01	1,957	$1,878.82	$0.00	$0.00	$306.07	$8,582			
A11.11.1.3			Scarify surface	62,765	m2	C.2.16	0.0	$0.03	$2,180.95	$0.07	$4,121	$0.02	1,115	$1,070.73	$0.00	$0.00	$0.12	$7,373			
A11.11.2.1		Revegetate	Seed/Fertilize: broadcast seeding	11.1	ha	C.6.01	366.5	$1,029.50	$11,389.08	$401.72	$4,444	$232.94	2,684	$2,576.99	$1,005.72	$11,125.98	$2,669.89	$29,536			
A11.11.2.2			Tree seedling application (1,000 stems/ha)	11.1	ha	C.6.06	567.0	$1,832.33	$20,270.56	$25.98	$287	$15.00	173	$165.94	$469.71	$5,196.26	$2,343.03	$25,920			
A11.12	*Pelly Laydown Area*																		$147,962		
A11.12.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	107	hrs	C.2.13	107.4	$44.39	$4,767.52	$194.68	$20,910	$67.01	7,497	$7,197.08	$0.00	$0.00	$306.07	$32,874			
A11.12.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	25,855	m3	R.058	380.7	$0.65	$16,779.44	$2.11	$54,545	$0.69	18,585	$17,841.31	$0.00	$0.00	$3.45	$89,165			
A11.12.3.1		Revegetate	Seed/Fertilize: broadcast seeding	5.17	ha	C.6.01	171.3	$1,029.50	$5,323.56	$401.72	$2,077	$232.94	1,255	$1,204.55	$1,005.72	$5,200.58	$2,669.89	$13,806			
A11.12.3.2			Tree seedling application (1,000 stems/ha)	5.17	ha	C.6.06	265.0	$1,832.33	$9,474.99	$25.98	$134	$15.00	81	$77.57	$469.71	$2,428.87	$2,343.03	$12,116			
A11.13	*W15 Sump Area Laydown*																		$36,022		
A11.13.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	9	hrs	C.2.13	8.8	$44.39	$390.13	$194.68	$1,711	$67.01	613	$588.94	$0.00	$0.00	$306.07	$2,690			
A11.13.1.2			Slopes: Re-grade to 3H:1V	1	hrs	C.2.13	1.2	$44.39	$55.15	$194.68	$242	$67.01	87	$83.26	$0.00	$0.00	$306.07	$380			
A11.13.1.3			Scarify surface	9,484	m2	C.2.16	0.0	$0.03	$329.54	$0.07	$623	$0.02	169	$161.79	$0.00	$0.00	$0.12	$1,114			
A11.13.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	6,974	m3	R.065	106.1	$0.67	$4,675.54	$2.18	$15,199	$0.71	5,179	$4,971.43	$0.00	$0.00	$3.56	$24,846			
A11.13.3.1		Revegetate	Seed/Fertilize: broadcast seeding	1.39	ha	C.6.01	46.2	$1,029.50	$1,435.97	$401.72	$560	$232.94	338	$324.92	$1,005.72	$1,402.80	$2,669.89	$3,724			
A11.13.3.2			Tree seedling application (1,000 stems/ha)	1.39	ha	C.6.06	71.5	$1,832.33	$2,555.78	$25.98	$36	$15.00	22	$20.92	$469.71	$655.16	$2,343.03	$3,268			
A12	**Waste Disposal**																			**$452,923**	
A12.1	*Hydrocarbon contaminated soils*																		$66,551		
A12.1.1.1		Excavate HC contaminated	Complete testing for contaminated soils	48	test pits	C.3.03	24.0	$22.24	$1,067.53	$2.89	$139	$2.92	146	$139.97	$204.00	$9,792.00	$232.05	$11,138			
A12.1.1.2			Excavate and haul contaminated soils to on-site landfarm facility	726	m3	R.038	26.0	$1.58	$1,144.07	$3.17	$2,298	$0.96	723	$693.78	$0.00	$0.00	$5.70	$4,136			
A12.1.2.1		Construct landfarm	Existing Facility assumed to have sufficient capacity	0	m3																
A12.1.3.1		Operate landfarm	Aerate contaminated soils (mix)	12	event	C.3.04	270.0	$999.20	$11,990.37	$923.10	$11,077	$333.60	4,170	$4,003.20	$0.00	$0.00	$2,255.90	$27,071			
A12.1.3.2			Annual confirmation sampling	3	years	C.3.05	48.0	$711.69	$2,135.06	$92.63	$278	$93.31	292	$279.94	$1,224.00	$3,672.00	$2,121.63	$6,365			
A12.1.4.1		Close landfarm	Remove soils and collect liner	726	m3	R.001	22.9	$1.39	$1,007.57	$3.03	$2,197	$0.90	680	$652.50	$0.00	$0.00	$5.31	$3,857			
A12.1.4.2			Cut/fold liner and place into waste disposal container	4,265	m2	C.1.15	49.6	$0.43	$1,850.89	$0.19	$830	$0.07	308	$295.29	$0.00	$0.00	$0.70	$2,976			
A12.1.4.3			Regrade area	4,265	m2	C.2.11	42.7	$0.44	$1,893.14	$1.28	$5,472	$0.35	1,567	$1,504.69	$0.00	$0.00	$2.08	$8,870			
A12.1.4.4			Revegetation: Seed/Fertilizer: broadcast seeding	0.43	ha	C.6.01	14.1	$1,029.50	$439.08	$401.72	$171	$232.94	103	$99.35	$1,005.72	$428.94	$2,669.89	$1,139			
A12.1.4.5			Tree seedling application (1,000 stems/ha)	0.43	ha	C.6.06	21.9	$1,832.33	$781.49	$25.98	$11	$15.00	7	$6.40	$469.71	$200.33	$2,343.03	$999			
	Metal Contaminated soils																		$159,118		
A12.2.1.1		Crusher area	Delineate contaminated soil areas	197	test pits	C.3.01	98.5	$22.24	$4,381.32	$2.89	$570	$2.92	598	$574.45	$265.20	$52,244.40	$293.25	$57,770			
A12.2.1.2			Load, haul, dump contaminated soils to underground	2,949	m3	R.039	102.1	$1.53	$4,499.64	$3.33	$9,810	$0.99	3,035	$2,913.97	$0.00	$0.00	$5.84	$17,224			
A12.2.1.3			Complete confirmation testing	197	ea	C.3.02	32.8	$7.43	$1,463.48	$1.33	$261	$0.00	0	$0.00	$13.26	$2,612.22	$22.01	$4,337			
A12.2.2.1		Mill Area	Delineate contaminated soil areas	198	test pits	C.3.01	99.0	$22.24	$4,403.56	$2.89	$573	$2.92	601	$577.37	$265.20	$52,509.60	$293.25	$58,064			
A12.2.2.2			Load, haul, dump contaminated soils to underground	2,973	m3	R.039	103.0	$1.53	$4,536.49	$3.33	$9,890	$0.99	3,060	$2,937.83	$0.00	$0.00	$5.84	$17,365			
A12.2.2.3			Complete confirmation testing	198	ea	C.3.02	33.0	$7.43	$1,470.91	$1.33	$262	$0.00	0	$0.00	$13.26	$2,625.48	$22.01	$4,359			
A12.3	*Solid Waste Landfill*																		$217,134		
A12.3.1.1		Construction landfill	Place waste from site facilities	12,920	LCM	C.2.07	550.6	$1.89	$24,440.80	$8.30	$107,194	$2.86	38,433	$36,895.99	$0.00	$0.00	$13.04	$168,531			
A12.3.1.2			Place fill to minimize voids in the debris	3,371	m3	R.001	106.2	$1.39	$4,679.67	$3.03	$10,203	$0.90	3,157	$3,030.55	$0.00	$0.00	$5.31	$17,913			
A12.3.2.1		Close landfill	Soil Cover: Load haul, dump spread, compact (0.6m)	5,024	m3	R.002	190.0	$1.61	$8,100.50	$3.21	$16,144	$0.99	5,196	$4,988.33	$0.00	$0.00	$5.82	$29,233			
A12.3.2.2			Revegetation: Seed/Fertilizer: broadcast seeding	0.55	ha	C.6.01	18.1	$1,029.50	$562.06	$401.72	$219	$232.94	132	$127.18	$1,005.72	$549.07	$2,669.89	$1,458			
A12.4	*Hazardous Material Off-Site Disposal*																		$10,120		
A12.4.1.1		Hazardous Waste	From Building Demo: Transport off-site for disposal	2	trip	C.7.04	14.5	$351.24	$634.94	$303.64	$549	$113.33	213	$204.86	$0.00	$0.00	$768.21	$1,389			
A12.4.3.3			Hazardous materials: Disposal and Tipping fees.	1	ls	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0	$0			
A12.4.2.1		Reagents	Reagents: Transport to Whitehorse	25	hrs	C.1.22	99.3	$159.30	$3,955.90	$95.83	$2,380	$26.93	697	$668.71	$0.00	$0.00	$282.05	$7,004			
A12.4.2.2			Reagents: Disposal and tipping fees	2.2	trips	C.7.04	18.0	$351.24	$789.50	$303.64	$682	$113.33	265	$254.73	$0.00	$0.00	$768.21	$1,727			
A13	**Surface Water Conveyance**																			**$3,181,261**	
A13.1	*Channel A - W-15 to Main Pit*																		$207,036		
A13.1.1.1		Excavate channel	Load, haul, dump locally	40,300	m3	R.902	700.5	$0.77	$31,093.04	$1.64	$66,199	$0.46	19,351	$18,576.93	$0.00	$0.00	$2.88	$115,869			
A13.1.2.1		Place channel materials	Geotextile: Supply and place	7,785	m3	C.5.08	221.4	$1.09	$8,520.21	$0.11	$854	$0.11	897	$860.72	$2.16	$16,802.77	$3.47	$27,038			
A13.1.2.2			Rip-rap (angular, high quality): Screen and stockpile	3,855	m3	C.2.15	0.0	$2.54	$9,777.29	$6.06	$23,344	$1.86	7,462	$7,163.19	$0.00	$0.00	$10.45	$40,285			
A13.1.2.3			Rip-rap: Load, haul, dump	3,855	m3	R.018	102.2	$1.17	$4,501.31	$2.55	$9,814	$0.76	3,037	$2,915.04	$0.00	$0.00	$4.47	$17,230			
A13.1.2.4			Rip-rap: Place and secure	3,855	m3	R.904	41.9	$0.48	$1,860.68	$0.91	$3,507	$0.32	1,299	$1,247.51	$0.00	$0.00	$1.72	$6,615			
A13.2	*Channel B - W-35 to Area 2 Pit*																		$198,304		
A13.2.1.1		Excavate channel	Load, haul, dump locally	48,038	m3	R.902	835.0	$0.77	$37,063.29	$1.64	$78,910	$0.46	23,067	$22,143.92	$0.00	$0.00	$2.88	$138,117			
A13.2.2.1		Place channel materials	Geotextile: Supply and place	4,682	m2	C.5.08	133.1	$1.09	$5,124.06	$0.11	$514	$0.11	539	$517.64	$2.16	$10,105.20	$3.47	$16,261			
A13.2.2.2			Rip-rap (angular, high quality): Screen and stockpile	2,554	m3	C.2.15	0.0	$2.54	$6,477.46	$6.06	$15,465	$1.86	4,943	$4,745.61	$0.00	$0.00	$10.45	$26,689			
A13.2.2.3			Rip-rap: Load, haul, dump	2,554	m3	R.020	76.2	$1.32	$3,358.44	$2.87	$7,322	$0.85	2,266	$2,174.93	$0.00	$0.00	$5.03	$12,855			
A13.2.2.4			Rip-rap: Place and secure	2,554	m3	R.904	27.8	$0.48	$1,232.70	$0.91	$2,323	$0.32	861	$826.47	$0.00	$0.00	$1.72	$4,382			
A13.3	*Channel C - Main Pit to Main Access Road*																		$125,034		
A13.3.2.1		Excavate channel	Load, haul, dump locally	8,656	m3	R.902	150.5	$0.77	$6,678.68	$1.64	$14,219	$0.46	4,157	$3,990.26	$0.00	$0.00	$2.88	$24,888			
A13.3.3.1		Intake structure	Excavate soils to competent foundation materials	2,450	Bm3	R.902	42.6	$0.77	$1,890.07	$1.64	$4,024	$0.46	1,176	$1,129.25	$0.00	$0.00	$2.88	$7,043			
A13.3.3.2			Supply an install precast-concrete intake wingwall intake structure	1	ls	C.5.12	140.0	$5,946	$5,945.57	$3,583	$3,583	$643.84	671	$643.84	$37,000	$37,000.00	$47,172	$47,172			
A13.3.3.6			Backfill and compact around structure	2,450	Cm3	C.2.04	237.3	$2.58	$6,323.25	$3.01	$7,367	$1.03	2,638	$2,532.57	$0.00	$0.00	$6.62	$16,223			
A13.3.4.1		Place channel materials	Geotextile: Supply and place	3,171	m2	C.5.08	90.2	$1.09	$3,470.31	$0.11	$348	$0.11	365	$350.57	$2.16	$6,843.83	$3.47	$11,013			
A13.3.4.2			Rip-rap (angular, high quality): Screen and stockpile	1,037	m3	C.2.15	0.0	$2.54	$2,631.28	$6.06	$6,282	$1.86	2,008	$1,927.76	$0.00	$0.00	$10.45	$10,841			
A13.3.4.3			Rip-rap: Load, haul, dump	1,037	m3	R.022	36.4	$1.54	$1,600.27	$3.33	$3,452	$0.98	1,064	$1,021.09	$0.00	$0.00	$5.85	$6,073			
A13.3.4.4			Rip-rap: Place and secure	1,037	m3	R.904	11.3	$0.48	$500.75	$0.91	$944	$0.32	350	$335.73	$0.00	$0.00	$1.72	$1,780			
A13.4	*Channel D - Area 2 Pit to Channel E*																		$195,636		
A13.4.1.1		Excavate channel	Load, haul, dump locally	17,840	m3	R.902	310.1	$0.77	$13,763.89	$1.64	$29,304	$0.46	8,566	$8,223.41	$0.00	$0.00	$2.88	$51,291			
A13.4.2.1		Intake structure	Excavate soils to competent foundation materials	4,453	Bm3	R.902	77.4	$0.77	$3,435.48	$1.64	$7,314	$0.46	2,138	$2,052.57	$0.00	$0.00	$2.88	$12,802			
A13.4.2.2			Supply an install precast-concrete intake wingwall intake structure	1	ls	C.5.12	140.0	$5,946	$5,945.57	$3,583	$3,583	$643.84	671	$643.84	$37,000	$37,000.00	$47,172	$47,172			
A13.4.2.6			Backfill and compact around structure	4,453	Cm3	C.2.04	431.4	$2.58	$11,493.42	$3.01	$13,390	$1.03	4,795	$4,603.31	$0.00	$0.00	$6.62	$29,487			
A13.4.3.1		Place channel materials	Geotextile: Supply and place	4,910	m2	C.5.08	139.6	$1.09	$5,373.23	$0.11	$539	$0.11	565	$542.81	$2.16	$10,596.59	$3.47	$17,051			
A13.4.3.2			Rip-rap (angular, high quality): Screen and stockpile	2,199	m3	C.2.15	0.0	$2.54	$5,578.78	$6.06	$13,320	$1.86	4,258	$4,087.21	$0.00	$0.00	$10.45	$22,986			
A13.4.3.3			Rip-rap: Load, haul, dump	2,199	m3	R.020	65.7	$1.32	$2,892.50	$2.87	$6,306	$0.85	1,951	$1,873.18	$0.00	$0.00	$5.03	$11,072			
A13.4.3.4			Rip-rap: Place and secure	2,199	m3	R.904	23.9	$0.48	$1,061.68	$0.91	$2,001	$0.32	741	$711.81	$0.00	$0.00	$1.72	$3,774			
A13.5	*Channel E - Main Access Road to toe of MVFE*																		$739,690		
A13.5.1.1		Construct channel	Bulk fill to grade: Load, haul, dump, spread	29,750	m3																Vol. from AutoCAD3D; Vol. placed as part of excavation of ren
A13.5.1.2			Excavate/form Channel: Load, haul, dump locally	46,386	m3	R.001	1,461.8	$1.39	$64,402.28	$3.03	$140,409	$0.90	43,445	$41,706.89	$0.00	$0.00	$5.31	$246,519			
A13.5.2.1		Place channel materials	Geotextile: Supply and place	27,307	m2	C.5.08	776.5	$1.09	$29,885.47	$0.11	$2,997	$0.11	3,145	$3,019.05	$2.16	$58,937.36	$3.47	$94,839			
A13.5.2.2			Rip-rap (angular, high quality): Screen and stockpile	20,558	m3	C.2.15	0.0	$2.54	$52,143.07	$6.06	$124,496	$1.86	39,794	$38,201.83	$0.00	$0.00	$10.45	$214,841			
A13.5.2.3			Rip-rap: Load, haul, dump	20,558	m3	R.025	845.7	$1.81	$37,226.38	$3.91	$80,296	$1.16	24,743	$23,753.12	$0.00	$0.00	$6.87	$141,275			
A13.5.2.4			Rip-rap: Place and secure	20,558	m3	R.904	223.6	$0.48	$9,923.17	$0.91	$18,701	$0.32	6,930	$6,653.05	$0.00	$0.00	$1.72	$35,277			
A13.5.3.1		Stilling Basin/Energy Dissip	Excavate basin (spoil locally)	645	m3	R.902	11.2	$0.77	$497.87	$1.64	$1,060	$0.46	310	$297.46	$0.00	$0.00	$2.88	$1,855			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A13.5.3.2			Geotextile: Supply and place	540	m2	C.5.08	15.4	$1.09	$591.07	$0.11	$59	$0.11	62	$59.71	$2.16	$1,165.65	$3.47	$1,876			
A13.5.3.3			Bedding layer: Screen and stockpile	131	m3	C.2.01	2.7	$0.96	$125.01	$2.11	$276	$0.44	60	$57.55	$0.00	$0.00	$3.51	$459			
A13.5.3.4			Bedding layer: Load, haul, dump and place	131	m3	R.028	7.7	$2.59	$337.69	$5.78	$755	$1.69	229	$220.12	$0.00	$0.00	$10.05	$1,312			
A13.5.3.5			Rip-rap (at inlet and outlet): Screen and stockpile	75	m3	C.2.15	0.0	$2.54	$190.23	$6.06	$454	$1.86	145	$139.37	$0.00	$0.00	$10.45	$784			
A13.5.3.6			Rip-rap (at inlet and outlet): Load, haul, dump	75	m3	R.027	3.2	$1.86	$139.15	$3.97	$298	$1.17	91	$87.84	$0.00	$0.00	$7.00	$525			
A13.5.3.7			Rip-rap (at inlet and outlet): Place and secure	75	m3	R.904	0.8	$0.48	$36.20	$0.91	$68	$0.32	25	$24.27	$0.00	$0.00	$1.72	$129			
A13.6	*Minto Creek Wetland By-Pass Channel*																		$231,721		
A13.6.1.1		Excavate channel	Load, haul, dump locally	29,668	m3	R.902	515.7	$0.77	$22,889.69	$1.64	$48,733	$0.46	14,246	$13,675.73	$0.00	$0.00	$2.88	$85,299			
A13.6.2.1		Place channel materials	Geotextile: Supply and place	12,212	m2	C.5.08	347.3	$1.09	$13,364.91	$0.11	$1,340	$0.11	1,406	$1,350.13	$2.16	$26,357.04	$3.47	$42,412			
A13.6.2.2			Rip-rap (angular, high quality): Screen and stockpile	5,950	m3	C.2.15	0.0	$2.54	$15,091.24	$6.06	$36,032	$1.86	11,517	$11,056.37	$0.00	$0.00	$10.45	$62,179			
A13.6.2.3			Rip-rap: Load, haul, dump	5,950	m3	R.001	187.5	$1.39	$8,260.73	$3.03	$18,010	$0.90	5,573	$5,349.64	$0.00	$0.00	$5.31	$31,620			Assumes WSP rip-rap can be re-used
A13.6.2.4			Rip-rap: Place and secure	5,950	m3	R.904	64.7	$0.48	$2,871.96	$0.91	$5,412	$0.32	2,006	$1,925.52	$0.00	$0.00	$1.72	$10,210			
A13.7	*Allowance: Minto North Spillway*																		$83,528		
A13.7.1.1		Excavate channel	Load, haul, dump locally	13,035	m3	R.902	226.6	$0.77	$10,056.98	$1.64	$21,412	$0.46	6,259	$6,008.67	$0.00	$0.00	$2.88	$37,477			
A13.7.2.1		Place channel materials	Geotextile: Supply and place	4,136	m2	C.5.08	117.6	$1.09	$4,526.14	$0.11	$454	$0.11	476	$457.23	$2.16	$8,926.04	$3.47	$14,363			
A13.7.2.2			Rip-rap (angular, high quality): Screen and stockpile	1,943	m3	C.2.15	0.0	$2.54	$4,929.10	$6.06	$11,769	$1.86	3,762	$3,611.23	$0.00	$0.00	$10.45	$20,309			
A13.7.2.3			Rip-rap: Load, haul, dump	1,943	m3	R.024	47.7	$1.08	$2,101.36	$2.36	$4,581	$0.70	1,418	$1,360.84	$0.00	$0.00	$4.14	$8,044			Assumes WSP rip-rap can be re-used
A13.7.2.4			Rip-rap: Place and secure	1,943	m3	R.904	21.1	$0.48	$938.04	$0.91	$1,768	$0.32	655	$628.91	$0.00	$0.00	$1.72	$3,335			
A13.8	*Secondary Catchment Channels*																		$775,342		
A13.8.1.1		Southwest Dump	Excavate mild-graded channels (good access), spoil locally	10,787	Bm3	R.902	187.5	$0.77	$8,322.78	$1.64	$17,720	$0.46	5,180	$4,972.55	$0.00	$0.00	$2.88	$31,015			
A13.8.1.2			Excavate steep-graded channels (poor access), spoil locally	6,640	Bm3	R.903	131.0	$0.88	$5,815.42	$1.86	$12,381	$0.52	3,619	$3,474.49	$0.00	$0.00	$3.26	$21,671			
A13.8.1.3			Geotextile: Supply and place	25,206	m2	C.5.08	716.7	$1.09	$27,585.63	$0.11	$2,766	$0.11	2,903	$2,786.72	$2.16	$54,401.82	$3.47	$87,541			
A13.8.1.4			Bedding layer: Screen and stockpile	2,054	Cm3	C.2.01	41.8	$0.96	$1,967.27	$2.11	$4,343	$0.44	943	$905.70	$0.00	$0.00	$3.51	$7,216			
A13.8.1.5			Bedding layer: Load, haul, dump and place in steep areas	2,054	Cm3	R.032	72.2	$1.55	$3,176.63	$3.34	$6,852	$0.99	2,111	$2,026.92	$0.00	$0.00	$5.87	$12,055			
A13.8.1.6			Rip-rap (angular, high quality): Screen and stockpile	6,806	Cm3	C.2.15	0.0	$2.54	$17,263.99	$6.06	$41,219	$1.86	13,175	$12,648.20	$0.00	$0.00	$10.45	$71,131			
A13.8.1.7			Rip-rap: Load, haul, dump	6,806	m3	R.031	223.5	$1.45	$9,846.77	$3.15	$21,468	$0.94	6,642	$6,376.76	$0.00	$0.00	$5.54	$37,691			
A13.8.1.8			Rip-rap: Place in mild-graded channels (good access)	3,787	m3	R.904	41.2	$0.48	$1,828.10	$0.91	$3,445	$0.32	1,277	$1,225.66	$0.00	$0.00	$1.72	$6,499			
A13.8.1.9			Rip-rap: Place in steep-graded channels (poor access)	3,019	m3	R.905	74.5	$1.10	$3,308.57	$2.33	$7,044	$0.65	2,059	$1,976.75	$0.00	$0.00	$4.08	$12,329			
A13.8.2.1		Main Waste Dump	Excavate mild-graded channels (good access), spoil locally	6,477	Bm3	R.902	112.6	$0.77	$4,997.00	$1.64	$10,639	$0.46	3,110	$2,985.52	$0.00	$0.00	$2.88	$18,621			
A13.8.2.2			Excavate steep-graded channels (poor access), spoil locally	3,986	Bm3	R.903	78.6	$0.88	$3,490.67	$1.86	$7,432	$0.52	2,172	$2,085.55	$0.00	$0.00	$3.26	$13,008			
A13.8.2.3			Geotextile: Supply and place	15,132	m2	C.5.08	430.3	$1.09	$16,560.93	$0.11	$1,661	$0.11	1,743	$1,672.99	$2.16	$32,659.93	$3.47	$52,555			
A13.8.2.4			Bedding layer: Screen and stockpile	1,233	Cm3	C.2.01	25.1	$0.96	$1,180.84	$2.11	$2,607	$0.44	566	$543.64	$0.00	$0.00	$3.51	$4,331			
A13.8.2.5			Bedding layer: Load, haul, dump and place in steep areas	1,233	Cm3	R.030	34.8	$1.25	$1,535.65	$2.76	$3,401	$0.82	1,059	$1,016.39	$0.00	$0.00	$4.83	$5,953			
A13.8.2.6			Rip-rap (angular, high quality): Screen and stockpile	4,086	Cm3	C.2.15	0.0	$2.54	$10,364.11	$6.06	$24,745	$1.86	7,909	$7,593.11	$0.00	$0.00	$10.45	$42,702			
A13.8.2.7			Rip-rap: Load, haul, dump	4,086	m3	R.029	101.4	$1.09	$4,473.09	$2.42	$9,908	$0.72	3,084	$2,960.56	$0.00	$0.00	$4.24	$17,341			
A13.8.2.8			Rip-rap: Place in mild-graded channels (good access)	2,274	m3	R.904	24.7	$0.48	$1,097.59	$0.91	$2,068	$0.32	767	$735.89	$0.00	$0.00	$1.72	$3,902			
A13.8.2.9			Rip-rap: Place in steep-graded channels (poor access)	1,812	m3	R.905	44.7	$1.10	$1,985.96	$2.33	$4,228	$0.65	1,236	$1,186.53	$0.00	$0.00	$4.08	$7,401			
A13.8.3.1		Reclamation Overburden D	Excavate mild-graded channels (good access), spoil locally	4,886	Bm3	R.902	84.9	$0.77	$3,769.89	$1.64	$8,026	$0.46	2,346	$2,252.36	$0.00	$0.00	$2.88	$14,049			
A13.8.3.2			Excavate steep-graded channels (poor access), spoil locally	3,007	Bm3	R.903	59.3	$0.88	$2,633.47	$1.86	$5,607	$0.52	1,639	$1,573.40	$0.00	$0.00	$3.26	$9,814			
A13.8.3.3			Geotextile: Supply and place	11,416	m2	C.5.08	324.6	$1.09	$12,494.07	$0.11	$1,253	$0.11	1,315	$1,262.16	$2.16	$24,639.65	$3.47	$39,649			
A13.8.3.4			Bedding layer: Screen and stockpile	930	Cm3	C.2.01	18.9	$0.96	$890.86	$2.11	$1,967	$0.44	427	$410.14	$0.00	$0.00	$3.51	$3,268			
A13.8.3.5			Bedding layer: Load, haul, dump and place in steep areas	930	Cm3	R.034	34.7	$1.64	$1,527.43	$3.58	$3,330	$1.06	1,030	$989.17	$0.00	$0.00	$6.29	$5,847			
A13.8.3.6			Rip-rap (angular, high quality): Screen and stockpile	3,083	Cm3	C.2.15	0.0	$2.54	$7,819.00	$6.06	$18,669	$1.86	5,967	$5,728.47	$0.00	$0.00	$10.45	$32,216			
A13.8.3.7			Rip-rap: Load, haul, dump	3,083	m3	R.033	97.0	$1.39	$4,271.96	$3.02	$9,314	$0.90	2,882	$2,766.52	$0.00	$0.00	$5.30	$16,352			
A13.8.3.8			Rip-rap: Place in mild-graded channels (good access)	1,715	m3	R.904	18.7	$0.48	$828.06	$0.91	$1,561	$0.32	578	$555.18	$0.00	$0.00	$1.72	$2,944			
A13.8.3.9			Rip-rap: Place in steep-graded channels (poor access)	1,367	m3	R.905	33.8	$1.10	$1,498.27	$2.33	$3,190	$0.65	932	$895.16	$0.00	$0.00	$4.08	$5,583			
A13.8.4.1		DSTSF & MVFE	Excavate mild-graded channels (good access), spoil locally	7,378	Bm3	R.902	128.3	$0.77	$5,692.78	$1.64	$12,120	$0.46	3,543	$3,401.22	$0.00	$0.00	$2.88	$21,214			
A13.8.4.2			Excavate steep-graded channels (poor access), spoil locally	4,541	Bm3	R.903	89.6	$0.88	$3,976.72	$1.86	$8,467	$0.52	2,475	$2,375.94	$0.00	$0.00	$3.26	$14,819			
A13.8.4.3			Geotextile: Supply and place	17,239	m2	C.5.08	490.2	$1.09	$18,866.88	$0.11	$1,892	$0.11	1,985	$1,905.94	$2.16	$37,207.52	$3.47	$59,872			
A13.8.4.4			Bedding layer: Screen and stockpile	1,405	Cm3	C.2.01	28.6	$0.96	$1,345.27	$2.11	$2,970	$0.44	645	$619.34	$0.00	$0.00	$3.51	$4,935			
A13.8.4.5			Bedding layer: Load, haul, dump and place in steep areas	1,405	Cm3	R.028	82.5	$2.59	$3,633.88	$5.78	$8,121	$1.69	2,467	$2,368.79	$0.00	$0.00	$10.05	$14,124			
A13.8.4.6			Rip-rap (angular, high quality): Screen and stockpile	4,655	Cm3	C.2.15	0.0	$2.54	$11,807.22	$6.06	$28,191	$1.86	9,011	$8,650.38	$0.00	$0.00	$10.45	$48,648			
A13.8.4.7			Rip-rap: Load, haul, dump	4,655	m3	C.5.08	132.4	$1.09	$5,094.58	$0.11	$511	$0.11	536	$514.66	$2.16	$10,047.05	$3.47	$16,167			
A13.8.4.8			Rip-rap: Place in mild-graded channels (good access)	2,590	m3	R.904	28.2	$0.48	$1,250.42	$0.91	$2,356	$0.32	873	$838.35	$0.00	$0.00	$1.72	$4,445			
A13.8.4.9			Rip-rap: Place in steep-graded channels (poor access)	2,065	m3	R.905	51.0	$1.10	$2,262.48	$2.33	$4,817	$0.65	1,408	$1,351.75	$0.00	$0.00	$4.08	$8,431			
A13.8	*Tailings Diversion Ditch*																		$624,970		
A13.8.1.2		Regrade	Slopes: Re-grade to 3H:1V	28	hrs	C.2.13	28	$44.39	$1,226	$194.68	$5,376	$67.01	1,928	$1,850	$0.00	$0	$306.07	$8,452			
A13.8.2.1		Cover	Flat area cover material: Load, haul, dump spread	25,182	Cm3	R.066	562.3	$0.98	$24,763	$3.20	$80,473	$1.05	27,585	$26,482	$0.00	$0	$5.23	$131,718			
A13.8.2.2			Slope area cover material: Load, haul, dump along crest	18,049	Cm3	R.067	352.6	$0.86	$15,513	$2.84	$51,216	$0.95	17,921	$17,204	$0.00	$0	$4.65	$83,933			
A13.8.2.3			Slope area cover material: Spread down slope	14	hrs	C.2.13	14.0	$44.39	$621	$194.68	$2,724	$67.01	977	$937	$0.00	$0	$306.07	$4,282			
A13.8.3.1		Revegetate	Seed/Fertilize: tractor application	9	ha	C.6.01	286.4	$1,029.50	$8,901	$401.72	$3,473	$232.94	2,098	$2,014	$1,005.72	$8,696	$2,670	$23,085			
A13.8.3.2			Tree seedling application (1,000 stems/ha)	9	ha	C.6.06	443.1	$1,832.33	$15,843	$25.98	$225	$15.00	135	$130	$469.71	$4,061	$2,343	$20,258			
A13.8.4.1		Extend Channel to WSP	Clear and grub footprint	13,889	m2	C.2.03	52.8	$0.17	$2,344	$0.49	$6,774	$0.13	1,940	$1,863	$0.00	$0	$0.79	$10,981			
A13.8.4.2			Geotextile: Supply and place	13,889	m2	C.5.08	394.9	$1.09	$15,200	$0.11	$1,524	$0.11	1,600	$1,536	$2.16	$29,976	$3.47	$48,236			
A13.8.4.3			Rip-rap (angular, high quality): Screen and stockpile	13,169	m3	C.2.15	0.0	$2.54	$33,403	$6.06	$79,753	$1.86	25,492	$24,472	$0.00	$0	$10.45	$137,629			
A13.8.4.4			Rip-rap: Load, haul, dump	13,169	m3	R.035	618.4	$2.07	$27,207	$4.42	$58,233	$1.30	17,890	$17,175	$0.00	$0	$7.79	$102,615			
A13.8.4.5			Rip-rap: Place in steep-graded channels (poor access)	13,169	m3	R.905	325.1	$1.10	$14,432	$2.33	$30,726	$0.65	8,982	$8,622	$0.00	$0	$4.08	$53,780			
A14	**Water Treatment**																			**$483,085**	
A14.1	*Passive Treatment System - Minto Creek Valley*																		$397,994		
A14.1.1.1		Water Management	Install pump and pump around system	8	hrs	C.4.14	32.0	$120.43	$963.46	$20.53	$164	$6.80	57	$54.40	$0.00	$0.00	$147.76	$1,182			Assumed
A14.1.1.2			Maintain pump around system during wetland construction activities	1,440	hrs	C.4.13	144.0	$3.55	$5,117.32	$12.98	$18,691	$2.00	3,000	$2,880.00	$0.00	$0.00	$18.53	$26,689			Assumes wetlands constructed over 2 months
A14.1.2.1		Valley Fill placement	Load, haul, dump material from the WSP excavation	85,918	m3	n/a - coste	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A14.1.2.1			Load, haul dump valley fill material	0	m3	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			Fill from WSP excavation exceeds the required amount.
A14.1.3.1		Construct wetlands	Excavate cells: Load, haul dump, spoil locally	14,459	m3	R.902	251.3	$0.77	$11,155.61	$1.64	$23,751	$0.46	6,943	$6,665.06	$0.00	$0.00	$2.88	$41,572			
A14.1.3.2			Subgrade preparation: remove sharp objects, place fill as required	9,273	m2	C.2.18	317.0	$1.31	$12,142.93	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$1.31	$12,143			
A14.1.3.3			Geotextile: Supply and install	10,201	m2	C.5.08	290.1	$1.09	$11,163.84	$0.11	$1,120	$0.11	1,175	$1,127.78	$2.16	$22,016.30	$3.47	$35,428			
A14.1.3.4			HDPE liner: Supply and install	10,201	m2	C.5.11	1,187.0	$4.74	$48,319.12	$0.27	$2,749	$0.27	2,884	$2,769.07	$9.38	$95,723.03	$14.66	$149,560			
A14.1.3.5			Organics: Mulch organics, produce and stockpile	5,122	Cm3	C.2.10	577.1	$4.50	$23,023.50	$4.87	$24,922	$1.82	9,720	$9,331.25	$0.00	$0.00	$11.18	$57,277			
A14.1.3.6			Organics: Load, haul, dump, place in cell	5,122	Cm3	R.001	161.4	$1.39	$7,111.43	$3.03	$15,504	$0.90	4,797	$4,605.36	$0.00	$0.00	$5.31	$27,221			
A14.1.3.7			Plant wetland vegetation	0.9	ha	C.6.05	40.6	$1,443.27	$1,338.39	$307.03	$285	$180.00	174	$166.92	$1,159.23	$1,074.99	$3,089.53	$2,865			
A14.1.4.1		Conveyance channels	Excavate channels: Load, haul, dump locally	1,402	m3	R.902	24.4	$0.77	$1,081.90	$1.64	$2,303	$0.46	673	$646.39	$0.00	$0.00	$2.88	$4,032			
A14.1.4.2			Surface preparation: remove sharp objects, place fill as required	1,858	m2	C.2.18	63.5	$1.31	$2,432.63	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$1.31	$2,433			
A14.1.4.3			Geotextile: Supply and place	1,858	m2	C.5.08	52.8	$1.09	$2,033.17	$0.11	$204	$0.11	214	$205.39	$2.16	$4,009.63	$3.47	$6,452			
A14.1.4.4			HDPE Liner: Supply and install	1,858	m2	C.5.11	216.2	$4.74	$8,799.92	$0.27	$501	$0.27	525	$504.31	$9.38	$17,433.16	$14.66	$27,238			
A14.1.4.5			Bedding/Protection layer: Screen and stockpile	442	m3	C.2.01	9.0	$0.96	$423.54	$2.11	$935	$0.44	203	$194.99	$0.00	$0.00	$3.51	$1,554			
A14.1.4.6			Bedding/Protection layer: Load, haul, dump and place	442	m3	R.001	13.9	$1.39	$614.04	$3.03	$1,339	$0.90	414	$397.65	$0.00	$0.00	$5.31	$2,350			Assumes WSP fill can be re-used
A14.1.5.1		Operation and maintenance	System to be started Yr. 1 post-closure	0	yr	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A14.2	*Southwest Wetlands*																		$15,944		
A14.2.1		Berm construction	Construct 3 berms across valley	3,000	m3	R.001	94.5	$1.39	$4,165.24	$3.03	$9,081	$0.90	2,810	$2,697.41	$0.00	$0.00	$5.31	$15,944			Assumed
A14.2.2		Outlet structures	Excavate outlet channels through berms	1	ls																Allowance

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A14.3	*Decommission/demolish Water Treatment System*																		$69,147		
A14.3.1.1		Remove equipment	Small equipment: dismantle and prep for transport	17	hrs	C.1.13	67.1	$168.00	$2,819.22	$86.19	$1,446	$14.53	254	$243.80	$0.00	$0.00	$268.71	$4,509			
A14.3.1.2			Large equipment (crane req'd): dismantle and prep for transport	29	hrs	C.1.14	175.7	$269.89	$7,903.93	$167.56	$4,907	$20.53	626	$601.18	$0.00	$0.00	$457.98	$13,412			
A14.3.2.1		Prepare for demolition	Reagents: Load for transport	25	hrs	C.1.22	99.3	$159.30	$3,955.90	$95.83	$2,380	$26.93	697	$668.71	$0.00	$0.00	$282.05	$7,004			
A14.3.2.2			Reagents: Transport to Whitehorse	2.2	trips	C.7.04	18.0	$351.24	$789.50	$303.64	$682	$113.33	265	$254.73	$0.00	$0.00	$768.21	$1,727			
A14.3.2.3			Reagents: Disposal and tipping fees	1.0	ls	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A14.3.3.1		Demolition	Structural building demolition	163	tonnes	C.1.05	325.0	$69.36	$11,272.12	$139.08	$22,604	$33.84	5,729	$5,499.69	$0.00	$0.00	$242.28	$39,376			
A14.3.3.2			Other demolition: miscellaneous	15	m3	C.1.07	1.5	$3.74	$56.09	$2.07	$31	$0.74	12	$11.05	$0.00	$0.00	$6.55	$98			
A14.3.4.1		Waste disposal	Off-site disposal (re-usable equipment, etc.)	3	trips	C.7.07	27.1	$355.10	$1,202.32	$423.68	$1,435	$113.33	400	$383.71	$0.00	$0.00	$892	$3,021			
Subtotal Direct Costs - Active Closure							**114,724**		**$4,864,705**		**$8,474,076**			**$2,573,011**		**$1,132,685**				**$17,044,478**	
INDIRECT COSTS																					
B1	**Mobilization-Demobilization**																			**$324,077**	
B1.1		Mobilization	Year 0	1	LS												$162,039	$162,038.53			
B1.2		Demobilization	Year 3 (end of Active Closure)	1	LS												$144,406	$144,405.50			
B1.4		Demobilization	End of Passive closure	1	LS												$17,633	$17,633.03			
B2	**Transportation Costs**																			**$405,599**	
B2.1.1		Barge Operations		14	month												$10,832	$151,654			
		Staffing Bus trips during barge operation period		56	ea (one per week)												$678	$37,946			Turnaround labour costs included in labour build-up
B2.2.2		Air transport and airstrip operations		72	flights												$3,000	$216,000			Three flights per week
B3	**Site/Road Maintenance**																			**$757,001**	
B3.1	*Road Maintenance*																		$693,226		
B3.1.1		Water Truck	Assume needed 150 hrs per month	20	month												$12,153	$243,052			
B3.1.2		Grader	Assume needed 150 hrs per month (16H + Operator)	20	month												$22,509	$450,173			Includes Main Access Road throughout closure period.
B3.2	*Soil Erosion*																		$63,775		
B3.2.1		Erosion control allowance	Silt fencing: supply, install remove at strategic locations	7023.7	m	C.2.17	283.5	$3.23	$22,661.96	$1.93	$13,533	$0.68	4,941	$4,743.08	$0.92	$6,485.58	$6.75	$47,424			
B3.2.2			Supply and install erosion control matting	10000	m2	C.2.09	175.4	$0.66	$6,622.75	$0.09	$912	$0.05	548	$526.32	$0.83	$8,290.76	$1.64	$16,352			
B4	**Construction Support**																			**$5,054,174**	
B4.1	*Field Support Staff*																		$3,081,160		Turnaround labour costs included in labour build-up
B4.1.1		Mine Manager		20	month			(360) site working hours per month									$23,408	$468,170			
B4.1.2		Office/Camp manager		20	month			(360) site working hours per month									$22,267	$445,347			
B4.1.3		Security/Administrative Assistant		20	month			(360) site working hours per month									$16,600	$331,991			
B4.1.4		Foreman		20	month			(360) site working hours per month									$17,769	$355,383			
B4.1.5		Mechanic		20	month			(360) site working hours per month									$17,769	$355,383			
B4.1.6		Surveyor		20	month			(360) site working hours per month									$17,769	$355,383			
B4.1.7		Engineering technician	Material & QA/QC testing	10	month			(360) site working hours per month									$42,120	$421,200			Assumed needed 1/2 of time
B4.1.8		Medic/H&S supervisor		20	month			(360) site working hours per month									$17,415	$348,303			
B4.2	*Field support Vehicles*																		$821,700		
B4.2.1		Pick-up trucks (4 required)		80	month												$4,446	$355,668			
B4.2.2		Fuel truck		20	month												$4,720	$94,403			
B4.2.3		Mechanic service vehicle		20	month												$11,803	$236,069			
B4.2.4		Emergency transport vehicle		20	month												$4,519	$90,382			
B4.2.5		Passenger bus		20	month												$2,259	$45,177			
B4.3	*Field Support Equipment/Supplies*																		$1,151,314		
B4.3.1		Light Towers		20	month			2 # of Units									$3,356	$67,120			
B4.3.2		Material/Laboratory testing allowance		20	month												$1,000	$20,000			
B4.4		Construction, Environmental, and H&S Management Plans		1	LS												$25,000	$25,000			
B4.5		Office supplies		20	month												$1,000	$20,000			
B4.6		Communications		20	month												$1,000	$20,000			
B4.7		Misc. supplies		20	month												$500	$10,000			
B4.8		Camp costs	Includes catering and housekeeping	11,472	man-day												$80	$917,793			
B4.9		Power and heat		20	month												$3,570	$71,400			
B5	**QA and Project Management**																			**$1,668,691**	
B5.1		Project Manager		20	month			(160) site working hours per month									$9,897	$197,932			
B5.2		Design Engineer		20	month			(360) site working hours per month									$48,600	$972,000			
B5.3		Environmental Monitor		20	month			(360) site working hours per month									$16,046	$320,926			
B5.4		Engineering , Design, and Construction plans (included in Planning and Permitting Costs)															$0.00	$0			
B5.5		Field support vehicles (2 vehicles)		40	month												$4,446	$177,834			
Subtotal Indirect Costs - Active Closure																				**$8,209,543**	
CLOSURE IMPLEMENTATION COSTS - TOTAL																				**$25,254,020**	

Worksheet 5 - Year 0 Estimate - Schedule Details

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

A: Year 0 Site Access Schedule

Stage	Flight Months	Helicopter Months	Barge Months	Total Months	Month: Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	Comments/Notes
Interim Care	2	5	5	12	H	H	H	F	F	B	B	B	B	B	H	H	
SUB-TOTAL INTERIM CARE	**2**	**5**	**5**	**12**													
Active Closure 1	2	0	5	7				F	F	B	B	B	B	B			
Active Closure 2	2	0	5	7				F	F	B	B	B	B	B			
Active Closure 3	2	0	4	6				F	F	B	B	B	B				
SUB-TOTAL ACTIVE CLOSURE	**6**	**0**	**14**	**20**													
Post-Closure 1	2	0	3	5				F	F	B	B	B					
SUB-TOTAL POST-CLOSURE 1	**2**	**0**	**3**	**5**													
Post-Closure 2	0	0	1	1						B							
SUB-TOTAL POST-CLOSURE 2	**0**	**0**	**1**	**1**													

B: Annual C&M Staffing Schedule - Interim Operations

#	Role		Months Required	Month: Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	Comments/Notes
Months Active at Site			**12**	**1**	**1**	**1**	**1**	**1**	**1**	**1**	**1**	**1**	**1**	**1**	**1**	
Administration/Office																
1	Mine Manager		4				0.5	0.5	0.5	0.5	0.5	0.5	0.5			
2	Office/Camp Mgr		6	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	
3	Payroll/Accounting/HR		4				0.5	0.5	0.5	0.5	0.5	0.5	0.5			
Water Treatment/Environmental Staffing																
1	Warehouse/Water Treatment Operator		5				1.0	1.0	1.0	1.0	1.0					
2	Environmental Manager		4				1.0	0.5	0.5	0.5	0.5	0.5	0.5			
3	Environmental Technician		7	0.5	0.5	0.5	1.0	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	
Operations																
4	Barge Operator		10						2.0	2.0	2.0	2.0	2.0			
5	Equipment operators		13				2.0	3.0	1.5	1.5	1.5	1.5	1.5			
6	HD Mechanic		2				0.5	0.3	0.3	0.3	0.25	0.25	0.25			
7	Tradesmen		2				0.5	0.3	0.3	0.3	0.25	0.25	0.5			
8	Labour/Helpers		10	1	1	1.0	1.0	1.0	0.5	0.5	0.5	0.5	0.5	1.0	1.0	
Camp/Support																
1	Cooks/Housecleaning etc.		0				0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0		Contract (accounted for in camp costs)
Other																
1	Visitors		2				0.25	0.25	0.25	0.25	0.25	0.25	0.25	0.0		
TOTAL			**66.5**	**2.0**	**2.0**	**2.0**	**8.8**	**8.3**	**8.3**	**8.3**	**8.3**	**7.3**	**7.5**	**2.0**	**2.0**	
		Annual man-days:	**1,995**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)
2. Orange highlighted cells are used for camp-man day calculations only.

C: Annual C&M Staffing Schedule - During Active Phase

#	Role		Months Required	Month: Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	Comments/Notes
Months Active at Site			**7**				**1**	**1**	**1**	**1**	**1**	**1**	**1**			
Water Treatment/Environmental Staffing																
1	Warehouse/Water Treatment Operator		5				1.0	1.0	1.0	1.0	1.0					
2	Environmental Manager		4				1.0	0.5	0.5	0.5	0.5	0.5	0.5			
3	Environmental Technician		4				1.0	0.5	0.5	0.5	0.5	0.5	0.5			
TOTAL			**13**	**0**	**0**	**0**	**3**	**2**	**2**	**2**	**2**	**1**	**1**	**0**	**0**	
		Annual man-days:	**390**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)

D: Annual C&M Staffing Schedule - Post Closure 1

#	Role		Months Required	Month												Comments/Notes
				Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	
Months Active at Site			**5**				**1**	**1**	**1**	**1**	**1**					
Administration/Office																
1	Mine Manager		3				0.5	0.5	0.5	0.5	0.5					
2	Office/Camp Mgr		3				1.0	0.5	0.5	0.5	0.5					
3	Payroll/Accounting/HR		1				0.25	0.25	0.25	0.25	0.25					
Water Treatment/Environmental Staffing																
1	Water Treatment Operator		2				0.5	0.5	0.3	0.3	0.3					
2	Environmental Manager		3				0.5	0.5	0.5	0.5	0.5					
3	Environmental Technician		3				0.5	0.5	0.5	0.5	0.5					
Operations																
4	Barge Operator		6						2.0	2.0	2.0					
5	Equipment operators		10				2.0	3.0	1.5	1.5	1.5					
6	HD Mechanic		2				1.0	0.25	0.25	0.25	0.25					
7	Tradesmen		2				1.0	0.25	0.25	0.25	0.25					
8	Labour/Helpers		2				1.0	0.25	0.25	0.25	0.5					
Camp/Support																
1	Cooks/Housecleaning etc.		0				0.0	0.0	0.0	0.0	0.0					Contract (accounted for in camp costs)
Other																
1	Visitors		1				0.25				0.25					
TOTAL			**36**	**0**	**0**	**0**	**9**	**7**	**7**	**7**	**7**	**0**	**0**	**0**	**0**	
	Annual man-days:		**1,073**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)
2. Orange highlighted cells are used for camp-man day calculations only.

E: Annual C&M Staffing Schedule - Post Closure 2

#	Role		Months Required	Month												Comments/Notes
				Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	
Months Active at Site			**1**						**1**							
Administration/Office																
1	Mine Manager		1						0.5							
2	Office/Camp Mgr		0						0.0							
3	Payroll/Accounting/HR		0						0.00							
Water Treatment/Environmental Staffing																
1	Passive Treatment Scientist		1						0.5							
2	Environmental Manager		0													
3	Environmental Technician		0													
Operations																
4	Barge Operator		1						1.0							
5	Equipment operators		1						1.0							
6	HD Mechanic		1						0.50							
7	Tradesmen		0													
8	Labour/Helpers		0						0.25							
Camp/Support																
1	Cooks/Housecleaning etc.		0						0.0							Contract (accounted for in camp costs)
Other																
1	Visitors		0						0.25							
TOTAL			**4**	**0**	**0**	**0**	**0**	**0**	**4**	**0**	**0**	**0**	**0**	**0**	**0**	
	Annual man-days:		**120**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)
2. Orange highlighted cells are used for camp-man day calculations only.

E: Annual C&M Staffing Schedule - Long-term perpetual maintenance year

#	Role		Months Required	Month												Comments/Notes
				Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	
Months Active at Site			**1**							**1**						
Administration/Office																
1	Project Manager/Foreman		1							1.0						
Water Treatment/Engineering Staffing																
1	Passive Treatment Scientist		1							1.0						
2	Site Engineer		1							1.0						
3	Environmental Technician		0													
Operations																
4	Barge Operator		1							1.0						Assumed required 1/2 the time (near mob and demob)
5	Equipment operators		4							4.0						
6	HD Mechanic		1							0.50						
7	Tradesmen		0													
8	Labour/Helpers		1							1.00						
Camp/Support																
1	Cooks/Housecleaning etc.		0													Contract (accounted for in camp costs)
Other																
1	Visitors		1							0.50						
TOTAL			**10**	**0**	**0**	**0**	**0**	**0**	**0**	**10**	**0**	**0**	**0**	**0**	**0**	
	Annual man-days:		**300**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)
2. Orange highlighted cells are used for camp-man day calculations only.

Worksheet 6- Year 0 Estimate - Planning and Permitting Costs

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

Planning And Permitting

WBS	Facility/Area	Task	Activity	Qnty	Unit	Total Unit Rate ($/unit)	Activity Total	Subtotal	Source / Comments
M1	**Planning and Permitting**								
M1.1	*Reclamation Research/Planning*							$400,000	
M1.1.1		Complete reclamation closure and research plan		2	yr	$200,000	$400,000		
M1.2	*Technical studies and investigations*							$58,500	
M1.2.1		Allowance for additional technical studies		1	ls	$58,500	$58,500		
M1.3	*Monitoring and Management Plans*							$209,158	
M1.3.1		Adaptive Mgmt Plans	Physcial, water quality, etc	1	ls	$15,000	$15,000		
M1.3.2		Revegetation plan		1	ls	$30,000	$30,000		
M1.3.3		Waste Management Plan	Water treatment, sludge, landfarm, etc.	1	ls	$50,000	$50,000		
M1.3.4		MMER Final Monitoring Study		1	ls	$114,158	$114,158		
M1.4	*Engineering, Design, and Construction Plans*							$852,224	
M1.4.1		Percentage of direct implementation costs		5%	of	$17,044,478	$852,224		
M1.5	*Permitting*							$0	
M1.5.1.1		Permit Staffing	Permitting Manager	0	ls	$0	$0		Staffing costs included in tasks above
M1.5.1.2			Environmental Manager	0	ls	$0	$0		
M1.5.1.3			Technical Consultants	0	ls	$0	$0		
TOTAL								**$1,519,882**	

Worksheet 7 - Peak Liability Estimate - Cashflow Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

NPV CALCULATION INPUTS

NOTE: Columns between Years 2047 and 2128 are hidden for printing purposes.

Short-Term Interest Rate (Interim Care/Implementation Costs)

##	Inflation Rate	Years in effect	Start Year	End Year
1	1.6%	100	2018	2118

Input Parameters

Scenario: Peak Liability - Mid 2019 - Interim Care starts 2020
Current Year: 2018 Year that the estimate is costed
Closure Year 1: 2022
Closure Period: 3 yrs
Post-Closure Year 1: 2025

Long-Term Monitoring/Maintenance NPV Discount Rates

##	Discount Rate	Years in effect	Start Year	End Year
1	2.35%	2	2018	2020
2	2.5%	3	2021	2024
3	2.8%	-	2025	

SCHEDULE

Implementation Schedule

			OPERATIONS		INTERIM CARE		CLOSURE			POST CLOSURE																							
		Closure Year:	Year -4	Year -3	Year -2	Year -1	Year 1	Year 2	Year 3	Year 4	Year 5	Year 6	Year 7	Year 8	Year 9	Year 10	Year 11	Year 12	Year 13	Year 14	Year 15	Year 16	Year 17	Year 18	Year 19	Year 20	Year 21	Year 22	Year 23	Year 24	Year 25	Year 26	Year 107	Year 108
	Area	Sum	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2128	2129
Direct Costs	Waste Dumps	100%					40%	60%																										
	Overburden Dumps	100%							100%																									
	Ore Stockpiles	100%					100%																											
	Open Pits	100%						50%	50%																									
	Underground Openings	100%					100%																											
	External Tailings Facilities	100%					50%	50%																										
	Roads	100%							100%																									
	Demolition	100%					10%	75%	15%																									
	Surface Infrastructure	100%					25%	50%	25%																									
	Water Detention Structures	100%						50%	50%																									
	Yards/Laydown Areas	100%					25%	25%	50%																									
	Waste Disposal	100%						12%	75%	2%	2%	2%	6%																					
	Surface Water Conveyance	100%					20%	30%	50%																									
	Water Treatment	100%							86%					14%																				
Indirect	Mobilization-Demobilization	100%					50%		45%					5%																				
	Transport Costs	100%					34%	33%	33%																									
	Road Maintenance	100%					34%	33%	33%																									
	Construction Support	100%					34%	33%	33%																									
	QA and Pro. Mgmt.	100%					34%	33%	33%																									

Monitoring Schedule

					1 - Interim	1 - Interim	2 - Active	2 - Active	2 - Active	3 - PC 1	3 - PC 1	3 - PC 1	3 - PC 1	3 - PC 1	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2										5 - Perpetual
		Post-Closure Year:	Year -7	Year -6	Year -5	Year -4	Year -3	Year -2	Year -1	Year 1	Year 2	Year 3	Year 4	Year 5	Year 6	Year 7	Year 8	Year 9	Year 10	Year 11	Year 12	Year 13	Year 14	Year 15	Year 16	Year 17	Year 18	Year 19	Year 20	Year 21	Year 22	Year 23	Year 104	Year 105
		Sum																																
Planning and permitting	Reclamation Research and Planning	100%			50%	50%																											
	Technical Studies and Investigations	100%			50%	50%																											
	Engineering, Design, and Construction Plans	100%			30%	40%	10%	10%	10%																									
	Monitoring and Management Plans	100%			50%	50%																												
	Permitting	100%			50%	50%																												
Monitoring	Water Quality Monitoring (surface and groundwater)				1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1									1	
	Sediment Monitoring				1	1	1	1	1	1		1		1					1					1									1	
	Biological Monitoring				1	1	1	1	1	1				1					1					1									1	
	Geotechnical Monitoring				1	1	1	1	1	1	1	1	1	1					1					1									1	
	Revegetation Monitoring						1	1	1	1	1	1	1	1					1					1									1	
	Annual Inspection Reporting				1	1	1	1	1	1	1	1	1	1					1					1									1	

Care and Maintenance Schedule

Water Treatment/Management	Active System				x	x	x	x	x	x	x	x	x	x	x	x		x	x	x	x	x	x	x										
	Passive System									x	x	x	x	x					x					x										x	
Reclamation Maintenance	Revegetation, cover repair, etc.									x	x	x	x	x																				x	
Site Maintenance					x	x	x	x	x	x	x	x	x	x					x					x										x	
On-site Management					x	x	x	x	x	x	x	x	x	x					x					x										x	
Transportation Costs					x	x	x	x	x	x	x	x	x	x					x					x										x	

UNDISCOUNTED CASH FLOW (TO 100 YEARS POST CLOSURE)

Implementation Costs

			Total	% of Total	Check Sum	Year -4	Year -3	Year -2	Year -1	Year 1	Year 2	Year 3	Year 4	Year 5	Year 6	Year 7	Year 8	Year 9	Year 10	Year 11	Year 12	Year 13	Year 14	Year 15	Year 16	Year 17	Year 18	Year 19	Year 20	Year 21	Year 22	Year 23	Year 24	Year 25	Year 26	Year 107	Year 108
	Area					2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2128	2129
Direct Costs	Waste Dumps		$6,893,723	25%	$6,893,723	$0	$0	$0	$0	$2,757,489	$4,136,234	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Overburden Dumps		$395,276	1%	$395,276	$0	$0	$0	$0	$0	$0	$395,276	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Ore Stockpiles		$3,331,999	12%	$3,331,999	$0	$0	$0	$0	$3,331,999	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Open Pits		$620,582	2%	$620,582	$0	$0	$0	$0	$0	$310,291	$310,291	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Underground Openings		$81,637	0%	$81,637	$0	$0	$0	$0	$81,637	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	External Tailings Facilities		$57,291	0%	$57,291	$0	$0	$0	$0	$28,646	$28,646	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Roads		$136,050	0%	$136,050	$0	$0	$0	$0	$0	$0	$136,050	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Demolition		$1,976,253	7%	$1,976,253	$0	$0	$0	$0	$197,625	$1,482,189	$296,438	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Surface Infrastructure		$169,527	1%	$169,527	$0	$0	$0	$0	$42,382	$84,763	$42,382	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Water Detention Structures		$786,952	3%	$786,952	$0	$0	$0	$0	$0	$393,476	$393,476	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Yards/Laydown Areas		$758,907	3%	$758,907	$0	$0	$0	$0	$189,727	$189,727	$379,454	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Waste Disposal		$452,923	2%	$452,923	$0	$0	$0	$0	$0	$54,828	$339,692	$11,145	$11,145	$11,145	$24,967	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Surface Water Conveyance		$3,287,496	12%	$3,287,496	$0	$0	$0	$0	$657,499	$986,249	$1,643,748	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Water Treatment		$483,085	2%	$483,085	$0	$0	$0	$0	$0	$0	$413,938	$0	$0	$0	$0	$69,147	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Indirect	Mobilization-Demobilization		$324,077	1%	$324,077	$0	$0	$0	$0	$162,039	$0	$144,406	$0	$0	$0	$0	$17,633	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Transport Costs		$419,142	1%	$419,142	$0	$0	$0	$0	$142,508	$138,317	$138,317	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Site/Road Maintenance		$791,662	3%	$791,662	$0	$0	$0	$0	$269,165	$261,249	$261,249	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Construction Support		$5,344,480	19%	$5,344,480	$0	$0	$0	$0	$1,817,123	$1,763,678	$1,763,678	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	QA and Pro. Mgmt.		$1,752,126	6%	$1,752,126	$0	$0	$0	$0	$595,723	$578,202	$578,202	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Subtotal - Closure Implementation Costs			**$28,063,189**		$28,063,189	**$0**	**$0**	**$0**	**$0**	**$10,273,563**	**$10,407,849**	**$7,236,595**	**$11,145**	**$11,145**	**$11,145**	**$24,967**	**$86,780**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

Monitoring Costs

			Total	% of Total	Check Sum	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2128	2129
1	Planning and	Reclamation Research and Planning	$400,000	10%	$400,000	$0	$0	$200,000	$200,000	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2	Permitting	Technical Studies and Investigations	$58,500	1%	$58,500	$0	$0	$29,250	$29,250	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3		Engineering, Design, and Construction Plans	$971,585	23%	$971,585	$0	$0	$291,476	$388,634	$97,159	$97,159	$97,159	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4		Monitoring and Management Plans	$209,158	5%	$209,158	$0	$0	$104,579	$104,579	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5		Permitting	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
1	Monitoring	Water Quality Monitoring	$1,403,014	33%	$1,403,014	$0	$0	$115,295	$115,295	$130,437	$130,437	$130,437	$51,987	$51,987	$51,987	$51,987	$51,987	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$0	$0	$0	$0	$0	$0	$0	$0	$0	$27,664
2		Sediment Monitoring	$64,372	2%	$64,372	$0	$0	$3,388	$3,388	$3,388	$3,388	$3,388	$0	$3,388	$0	$3,388	$0	$0	$0	$0	$0	$3,388	$0	$0	$0	$3,388	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$3,388
3		Biological Monitoring	$233,587	6%	$233,587	$0	$0	$12,294	$12,294	$12,294	$12,294	$12,294	$12,294	$0	$12,294	$0	$12,294	$0	$0	$0	$0	$12,294	$0	$0	$0	$12,294	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$12,294
4		Geotechnical Monitoring	$379,471	9%	$379,471	$0	$0	$12,409	$12,409	$12,409	$12,409	$12,409	$12,409	$12,409	$12,409	$12,409	$12,409	$0	$0	$0	$0	$23,216	$0	$0	$0	$23,216	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$23,216
5		Revegetation Monitoring	$200,071	5%	$200,071	$0	$0	$0	$0	$10,530	$10,530	$10,530	$10,530	$10,530	$10,530	$10,530	$10,530	$0	$0	$0	$0	$10,530	$0	$0	$0	$10,530	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$10,530
6		Annual Inspection Reporting	$286,620	7%	$286,620	$0	$0	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$23,900	$0	$0	$0	$23,900	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$23,900
Post Closure Costs Subtotal			**$4,206,378**		$4,206,378	**$0**	**$0**	**$771,063**	**$868,221**	**$268,589**	**$268,589**	**$268,589**	**$92,981**	**$77,299**	**$92,981**	**$77,299**	**$92,981**	**$27,220**	**$27,220**	**$27,220**	**$27,220**	**$100,548**	**$27,220**	**$27,220**	**$27,220**	**$27,220**	**$100,548**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$100,993**

Care and Maintenance Costs

			Total	% of Total	Check Sum	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2128	2129
1	Water Treatment	Active System	$7,981,012	36%	$7,981,012	$0.00	$0	$706,967	$706,967	$706,967	$706,967	$706,967	$706,967	$706,967	$706,967	$706,967	$706,967	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Passive System	$560,239	3%	$560,239	$0	$0	$0	$0	$0	$0	$0	$35,015	$35,015	$35,015	$35,015	$35,015	$0	$0	$0	$0	$35,015	$0	$0	$0	$0	$35,015	$0	$0	$0	$0	$0	$0	$0	$0	$0	$35,015
3	Reclamation Maintenance	(Cover/Reveg Repairs)	$1,834,053	8%	$1,834,053	$0	$0	$0	$0	$0	$0	$0	$366,811	$366,811	$366,811	$366,811	$366,811	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4	Site Maintenance		$2,478,551	11%	$2,478,551	$0	$0	$359,187	$359,187	$30,000	$30,000	$30,000	$99,414	$99,414	$99,414	$99,414	$99,414	$0	$0	$0	$0	$99,414	$0	$0	$0	$0	$99,414	$0	$0	$0	$0	$0	$0	$0	$0	$0	$108,253
5	On-site Management		$7,978,119	36%	$7,978,119	$0	$0	$1,124,816	$1,124,816	$310,135	$310,135	$310,135	$606,140	$606,140	$606,140	$606,140	$606,140	$0	$0	$0	$0	$79,830	$0	$0	$0	$0	$79,830	$0	$0	$0	$0	$0	$0	$0	$0	$0	$178,636
6	Transportation Costs		$1,297,725	6%	$1,297,725	$0	$0	$151,714	$151,714	$0	$0	$0	$64,628	$64,628	$64,628	$64,628	$64,628	$0	$0	$0	$0	$20,832	$0	$0	$0	$0	$20,832	$0	$0	$0	$0	$0	$0	$0	$0	$0	$69,943
Post Closure Costs Subtotal			**$22,129,700**		$22,129,700	**$0**	**$0**	**$2,342,683**	**$2,342,683**	**$1,047,102**	**$1,047,102**	**$1,047,102**	**$1,878,975**	**$1,878,975**	**$1,878,975**	**$1,878,975**	**$1,878,975**	**$91,135**	**$91,135**	**$91,135**	**$91,135**	**$326,226**	**$91,135**	**$91,135**	**$91,135**	**$91,135**	**$326,226**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$391,847**

Other Add-on Costs

			Total	% of Total	Check Sum	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2128	2129
1	Contingency	12% % of above costs	$6,527,912		$6,527,912	$0	$0	$373,650	$385,309	$1,390,710	$1,406,825	$1,026,274	$237,972	$236,090	$237,972	$237,749	$247,048	$14,203	$14,203	$14,203	$14,203	$51,213	$14,203	$14,203	$14,203	$14,203	$51,213	$0	$0	$0	$0	$0	$0	$0	$0	$0	$59,141
2	Worker's compensation	0% (Included in labour rates)	$0		$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3	Insurance	1.5% % of above labour costs	$224,359.88		$224,360	$0	$0	$12,842	$13,243	$47,798	$48,352	$35,272	$8,179	$8,114	$8,179	$8,171	$8,491	$488	$488	$488	$488	$1,760	$488	$488	$488	$488	$1,760	$0	$0	$0	$0	$0	$0	$0	$0	$0	$2,033
4	Bonding	3% % of Implem. & C&M costs	$1,505,786.67		$1,505,787	$0.00	$0	$70,280	$70,280	$339,620	$343,649	$248,511	$56,704	$56,704	$56,704	$57,118	$58,973	$2,734	$2,734	$2,734	$2,734	$9,787	$2,734	$2,734	$2,734	$2,734	$9,787	$0	$0	$0	$0	$0	$0	$0	$0	$0	$11,755
5	Contractor Profit	10% % of above costs	$5,439,927		$5,439,927	$0	$0	$311,375	$321,090	$1,158,925	$1,172,354	$855,229	$198,310	$196,742	$198,310	$198,124	$205,874	$11,835	$11,835	$11,835	$11,835	$42,677	$11,835	$11,835	$11,835	$11,835	$42,677	$0	$0	$0	$0	$0	$0	$0	$0	$0	$49,284
6	Contract Administration	5% % of above costs	$2,719,963		$2,719,963	$0	$0	$155,687	$160,545	$579,463	$586,177	$427,614	$99,155	$98,371	$99,155	$99,062	$102,937	$5,918	$5,918	$5,918	$5,918	$21,339	$5,918	$5,918	$5,918	$5,918	$21,339	$0	$0	$0	$0	$0	$0	$0	$0	$0	$24,642
Other Add-On Costs Subtotal			**$16,417,949**		$16,417,949	**$0**	**$0**	**$923,834**	**$950,468**	**$3,516,516**	**$3,557,356**	**$2,592,901**	**$600,320**	**$596,021**	**$600,320**	**$600,224**	**$623,322**	**$35,178**	**$35,178**	**$35,178**	**$35,178**	**$126,776**	**$35,178**	**$35,178**	**$35,178**	**$35,178**	**$126,776**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$146,855**

TOTAL			**$70,817,215**		$70,817,215	**$0**	**$0**	**$4,037,580**	**$4,161,372**	**$15,105,770**	**$15,280,896**	**$11,145,187**	**$2,583,420**	**$2,563,440**	**$2,583,420**	**$2,581,465**	**$2,682,058**	**$153,532**	**$153,532**	**$153,532**	**$153,532**	**$553,550**	**$153,532**	**$153,532**	**$153,532**	**$153,532**	**$553,550**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$639,695**

INFLATION FACTORED INTERIM CARE AND CLOSURE IMPLEMENTATION COSTS

	Area	Present Value	% of Total	Future Value	Year -4 2018	Year -3 2019	Year -2 2020	Year -1 2021	Year 1 2022	Year 2 2023	Year 3 2024	Year 4 2025	Year 5 2026	Year 6 2027	Year 7 2028	Year 8 2029	Year 9 2030	Year 10 2031	Year 11 2032	Year 12 2033	Year 13 2034	Year 14 2035	Year 15 2036	Year 16 2037	Year 17 2038	Year 18 2039	Year 19 2040	Year 20 2041	Year 21 2042	Year 22 2043	Year 23 2044	Year 24 2045	Year 25 2046	Year 26 2047	Year 107 2128	Year 108 2129
Direct Costs	Waste Dumps	$6,893,723	25%	$7,416,141	$0	$0	$0	$0	$2,938,249	$4,477,892	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Overburden Dumps	$395,276	1%	$434,773	$0	$0	$0	$0	$0	$0	$434,773	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Ore Stockpiles	$3,331,999	12%	$3,550,420	$0	$0	$0	$0	$3,550,420	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Open Pits	$620,582	2%	$677,218	$0	$0	$0	$0	$0	$335,922	$341,296	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Underground Openings	$81,637	0%	$86,989	$0	$0	$0	$0	$86,989	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	External Tailings Facilities	$57,291	0%	$61,535	$0	$0	$0	$0	$30,523	$31,012	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Roads	$136,050	0%	$149,644	$0	$0	$0	$0	$0	$0	$149,644	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Demolition	$1,976,253	7%	$2,141,259	$0	$0	$0	$0	$210,580	$1,604,620	$326,059	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Surface Infrastructure	$169,527	1%	$183,542	$0	$0	$0	$0	$45,160	$91,765	$46,617	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Water Detention Structures	$786,952	3%	$858,770	$0	$0	$0	$0	$0	$425,977	$432,793	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Yards/Laydown Areas	$758,907	3%	$824,932	$0	$0	$0	$0	$202,164	$205,398	$417,370	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Waste Disposal	$452,923	2%	$500,221	$0	$0	$0	$0	$0	$59,357	$373,635	$12,455	$12,654	$12,857	$29,262	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Surface Water Conveyance	$3,287,496	12%	$3,576,310	$0	$0	$0	$0	$700,600	$1,067,714	$1,807,996	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Water Treatment	$483,085	2%	$537,639	$0	$0	$0	$0	$0	$0	$455,300	$0	$0	$0	$0	$82,339	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Indirect	Mobilization-Demobilization	$324,077	1%	$352,492	$0	$0	$0	$0	$172,661	$0	$158,835	$0	$0	$0	$0	$20,997	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Transport Costs	$419,142	1%	$453,730	$0	$0	$0	$0	$151,850	$149,742	$152,138	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Road Maintenance	$791,662	3%	$856,991	$0	$0	$0	$0	$286,810	$282,828	$287,353	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Construction Support	$5,344,480	19%	$5,785,511	$0	$0	$0	$0	$1,936,240	$1,909,361	$1,939,910	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	QA and Pro. Mgmt.	$1,752,126	6%	$1,896,713	$0	$0	$0	$0	$634,774	$625,962	$635,977	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Subtotal - Closure Implementation Costs		**$28,063,189**		**$30,344,831**	**$0**	**$0**	**$0**	**$0**	**$10,947,020**	**$11,267,550**	**$7,959,697**	**$12,455**	**$12,654**	**$12,857**	**$29,262**	**$103,336**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

Monitoring Costs

			Present Value	% of Total	Future Value	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2128	2129
1	Planning and permitting	Reclamation Research and Planning	$400,000	10%	$416,206	$0	$0	$206,451	$209,754	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Technical Studies and Investigations	$58,500	1%	$60,870	$0	$0	$30,193	$30,677	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3		Engineering, Design, and Construction Plans	$971,585	23%	$1,024,044	$0	$0	$300,877	$407,589	$103,527	$105,184	$106,867	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4		Monitoring and Management Plans	$209,158	5%	$217,632	$0	$0	$107,952	$109,680	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5		Permitting	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
1	Monitoring	Water Quality Monitoring	$621,903	15%	$663,604	$0	$0	$119,014	$120,919	$138,988	$141,212	$143,471	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Sediment Monitoring	$16,940	0%	$18,055	$0	$0	$3,497	$3,553	$3,610	$3,668	$3,727	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3		Biological Monitoring	$61,470	1%	$65,516	$0	$0	$12,691	$12,894	$13,100	$13,310	$13,523	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4		Geotechnical Monitoring	$62,045	1%	$66,129	$0	$0	$12,809	$13,014	$13,222	$13,434	$13,649	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5		Revegetation Monitoring	$31,590	1%	$34,202	$0	$0	$0	$0	$11,220	$11,400	$11,582	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
6		Annual Inspection Reporting	$11,860	0%	$12,641	$0	$0	$2,449	$2,488	$2,527	$2,568	$2,609	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Post Closure Costs Subtotal			**$2,445,052**		**$2,578,899**	**$0**	**$0**	**$795,934**	**$910,566**	**$286,196**	**$290,775**	**$295,427**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

Care and Maintenance Costs

			Present Value	% of Total	Future Value	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2128	2129
1	Water Treatment	Active System	$3,534,833	16%	$3,767,499	$0	$0	$729,770	$741,447	$753,310	$765,363	$777,609	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Passive System	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3	Reclamation Maintenance	(Cover/Reveg Repairs)	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4	Site Maintenance		$808,374	4%	$844,921	$0	$0	$370,773	$376,706	$31,967	$32,478	$32,998	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5	On-site Management		$3,180,037	14%	$3,348,116	$0	$0	$1,161,098	$1,179,675	$330,466	$335,753	$341,125	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
6	Transportation Costs		$303,428	1%	$315,721	$0	$0	$156,608	$159,113	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Post Closure Costs Subtotal			**$7,826,673**		**$8,276,257**	**$0**	**$0**	**$2,418,249**	**$2,456,941**	**$1,115,742**	**$1,133,594**	**$1,151,732**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

Other Add-on Costs

			Present Value	% of Total	Future Value	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2128	2129
1	Contingency	12% % of above costs	$4,582,768		$4,923,531	$0	$0	$385,702	$404,101	$1,481,875	$1,523,030	$1,128,823	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2	Worker's compensation	0% (Included in labour rates)	$0		$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3	Insurance	2% % of above labour costs	$157,507		$169,218	$0	$0	$13,256	$13,889	$50,931	$52,346	$38,797	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4	Bonding	3% % of Implem. & C&M costs	$1,072,340		$1,153,516	$0	$0	$72,547	$73,708	$361,883	$372,034	$273,343	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5	Contractor Profit	10% % of above costs	$3,818,973		$4,102,942	$0	$0	$321,418	$336,751	$1,234,896	$1,269,192	$940,686	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
6	Contract Administration	5% % of above costs	$1,909,487		$2,051,471	$0	$0	$160,709	$168,375	$617,448	$634,596	$470,343	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Other Add-On Costs Subtotal			**$11,541,074**		$12,400,678	**$0**	**$0**	**$953,633**	**$996,824**	**$3,747,032**	**$3,851,198**	**$2,851,991**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**
TOTAL			**$49,875,988**		**$53,600,666**	**$0**	**$0**	**$4,167,817**	**$4,364,331**	**$16,095,990**	**$16,543,117**	**$12,258,847**	**$12,455**	**$12,654**	**$12,857**	**$29,262**	**$103,336**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

NET PRESENT VALUE CALCULATIONS - LONG TERM MONITORING AND MAINTENANCE

Monitoring Costs

			Undiscounted	% of Total	NPV - 2018	Year -4 2018	Year -3 2019	Year -2 2020	Year -1 2021	Year 1 2022	Year 2 2023	Year 3 2024	Year 4 2025	Year 5 2026	Year 6 2027	Year 7 2028	Year 8 2029	Year 9 2030	Year 10 2031	Year 11 2032	Year 12 2033	Year 13 2034	Year 14 2035	Year 15 2036	Year 16 2037	Year 17 2038	Year 18 2039	Year 19 2040	Year 20 2041	Year 21 2042	Year 22 2043	Year 23 2044	Year 24 2045	Year 25 2046	Year 26 2047	Year 107 2128	Year 108 2129
1	Monitoring	Water Quality Monitoring	$781,111	19%	$440,095								$44,829	$43,608	$42,420	$41,265	$40,141	$20,444	$19,888	$19,346	$18,819	$18,306	$17,808	$17,323	$16,851	$16,392	$15,945	$0	$0	$0	$0	$0	$0	$0	$0	$0	$1,350
2		Sediment Monitoring	$47,432	1%	$18,286								$2,921	$0	$2,765	$0	$2,616	$0	$0	$0	$0	$2,279	$0	$0	$0	$0	$1,985	$0	$0	$0	$0	$0	$0	$0	$0	$0	$165
3		Biological Monitoring	$172,117	4%	$66,354								$10,601	$0	$10,032	$0	$9,493	$0	$0	$0	$0	$8,268	$0	$0	$0	$0	$7,202	$0	$0	$0	$0	$0	$0	$0	$0	$0	$600
4		Geotechnical Monitoring	$317,426	8%	$119,080								$10,700	$10,409	$10,125	$9,850	$9,581	$0	$0	$0	$0	$15,614	$0	$0	$0	$0	$13,600	$0	$0	$0	$0	$0	$0	$0	$0	$0	$1,133
5		Revegetation Monitoring	$168,481	4%	$74,024								$9,080	$8,833	$8,592	$8,358	$8,131	$0	$0	$0	$0	$7,082	$0	$0	$0	$0	$6,169	$0	$0	$0	$0	$0	$0	$0	$0	$0	$514
6		Annual Inspection Reporting	$274,760	7%	$80,114								$2,045	$1,990	$1,935	$1,883	$1,831	$0	$0	$0	$0	$16,074	$0	$0	$0	$0	$14,001	$0	$0	$0	$0	$0	$0	$0	$0	$0	$1,166
Post Closure Costs Subtotal			**$1,761,326**		**$797,952**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$80,177**	**$64,839**	**$75,869**	**$61,355**	**$71,793**	**$20,444**	**$19,888**	**$19,346**	**$18,819**	**$67,623**	**$17,808**	**$17,323**	**$16,851**	**$16,392**	**$58,902**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$4,928**

Care and Maintenance Costs

			Undiscounted	% of Total	NPV - 2018	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2128	2129
1	Water Treatment	Active System	$4,446,180	20%	$3,492,933								$609,619	$593,015	$576,862	$561,150	$545,866	$68,451	$66,586	$64,773	$63,008	$61,292	$59,623	$57,999	$56,419	$54,882	$53,388	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Passive System	$560,239	3%	$246,147								$30,193	$29,371	$28,571	$27,793	$27,036	$0	$0	$0	$0	$23,549	$0	$0	$0	$0	$20,512	$0	$0	$0	$0	$0	$0	$0	$0	$0	$1,709
3	Reclamation Maintenance	(Cover/Reveg Repairs)	$1,834,053	8%	$1,497,671								$316,302	$307,687	$299,306	$291,154	$283,223	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4	Site Maintenance		$1,670,177	8%	$713,783								$85,725	$83,390	$81,119	$78,909	$76,760	$0	$0	$0	$0	$66,860	$0	$0	$0	$0	$58,238	$0	$0	$0	$0	$0	$0	$0	$0	$0	$5,282
5	On-site Management		$4,798,081	22%	$2,876,918								$522,676	$508,440	$494,591	$481,120	$468,016	$0	$0	$0	$0	$53,689	$0	$0	$0	$0	$46,765	$0	$0	$0	$0	$0	$0	$0	$0	$0	$8,717
6	Transportation Costs		$994,297	4%	$408,186								$55,729	$54,211	$52,735	$51,298	$49,901	$0	$0	$0	$0	$14,011	$0	$0	$0	$0	$12,204	$0	$0	$0	$0	$0	$0	$0	$0	$0	$3,413
Post Closure Costs Subtotal			**$14,303,027**		$9,235,638	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$1,620,245**	**$1,576,113**	**$1,533,184**	**$1,491,424**	**$1,450,802**	**$68,451**	**$66,586**	**$64,773**	**$63,008**	**$219,402**	**$59,623**	**$57,999**	**$56,419**	**$54,882**	**$191,106**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$19,120**

Other Add-on Costs

			Undiscounted	% of Total	NPV - 2018	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2128	2129
1	Contingency	12% % of above costs	$1,945,144		$1,217,816								$205,204	$198,036	$194,178	$188,712	$190,752	$10,667	$10,377	$10,094	$9,819	$34,443	$9,292	$9,039	$8,792	$8,553	$30,001	$0	$0	$0	$0	$0	$0	$0	$0	$0	$2,886
2	Worker's compensation	0% (Included in labour rates)	$0		$0								$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3	Insurance	2% % of above labour costs	$66,853		$41,856								$7,053	$6,806	$6,674	$6,486	$6,556	$367	$357	$347	$337	$1,184	$319	$311	$302	$294	$1,031	$0	$0	$0	$0	$0	$0	$0	$0	$0	$99
4	Bonding	3% % of Implem. & C&M costs	$433,446		$280,515								$48,896	$47,564	$46,268	$45,337	$45,534	$2,054	$1,998	$1,943	$1,890	$6,582	$1,789	$1,740	$1,693	$1,646	$5,733	$0	$0	$0	$0	$0	$0	$0	$0	$0	$574
5	Contractor Profit	10% % of above costs	$1,620,954		$1,014,847								$171,003	$165,030	$161,815	$157,260	$158,960	$8,890	$8,647	$8,412	$8,183	$28,702	$7,743	$7,532	$7,327	$7,127	$25,001	$0	$0	$0	$0	$0	$0	$0	$0	$0	$2,405
6	Contract Administration	5% % of above costs	$810,477		$507,423								$85,502	$82,515	$80,907	$78,630	$79,480	$4,445	$4,324	$4,206	$4,091	$14,351	$3,872	$3,766	$3,663	$3,564	$12,500	$0	$0	$0	$0	$0	$0	$0	$0	$0	$1,202
Other Add-On Costs Subtotal			**$4,876,874**		$3,062,457	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$517,657**	**$499,952**	**$489,842**	**$476,424**	**$481,282**	**$26,422**	**$25,702**	**$25,002**	**$24,321**	**$85,263**	**$23,014**	**$22,387**	**$21,778**	**$21,184**	**$74,266**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$7,166**
TOTAL			**$20,941,227**		**$13,096,047**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$2,218,079**	**$2,140,904**	**$2,098,895**	**$2,029,204**	**$2,003,876**	**$115,317**	**$112,176**	**$109,121**	**$106,148**	**$372,287**	**$100,445**	**$97,709**	**$95,048**	**$92,459**	**$324,274**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$31,214**
GRAND-TOTAL					**$66,696,713**	**$0**	**$0**	**$4,167,817**	**$4,364,331**	**$16,095,990**	**$16,543,117**	**$12,258,847**	**$2,230,534**	**$2,153,558**	**$2,111,752**	**$2,058,466**	**$2,107,212**	**$115,317**	**$112,176**	**$109,121**	**$106,148**	**$372,287**	**$100,445**	**$97,709**	**$95,048**	**$92,459**	**$324,274**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$31,214**

Worksheet 8 - Peak Liability Estimate - Annual C&M Costs

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

Summary of Annual Care and Maintenance Costs

		Annual Cost Per Phase							Number of Years				
		1 - Interim	2 - Active	3 - PC 1	4 - PC 2	5 - Perpetual	n/a	Total Cost	1 - Interim	2 - Active	3 - PC 1	4 - PC 2	5 - Perpetual
1	Active Water Treatment System	$706,967	$706,967	$706,967	$91,135	$0	$0	$7,981,012	2	3	5	10	0
2	Passive Treatment System	$0	$0	$35,015	$35,015	$35,015	$0	$560,239	2	3	5	2	9
3	Reclamation Maintenance	$0	$0	$366,811	$0	$0	$0	$1,834,053	2	3	5	2	9
4	Site Maintenance	$359,187	$30,000	$99,414	$99,414	$108,253	$0	$2,478,551	2	3	5	2	9
5	On-site Management	$1,124,816	$310,135	$606,140	$79,830	$178,636	$0	$7,978,119	2	3	5	2	9
6	Transportation Costs	$151,714	$0	$64,628	$20,832	$69,943	$0	$1,297,725	2	3	5	2	9

NOTES:
1. This table is used as a check to the NPV calculation spreadsheet.

Phase 1 - Annual Care & Maintenance Costs - Prior to Closure

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S1.1	**Water Treatment**															**$706,967**	
S1.1.1	*Active Treatment System*														$706,967		
S1.1.1.1		Operation and maintenance		1	yr						$511,807	$65,194	$38,831	$615,832			See details on 'Water Treatment' worksheet.
S1.1.1.2		Capital Replacement	annual allowance	1	yr						$86,932	$4,203	$0	$91,135			See details on 'Water Treatment' worksheet.
S1.2	**Onsite Management**															**$1,124,816**	
S1.2.1	*Field Support Staff*														$761,189		
S1.2.1.1		Mine Manager		4	months	360	site working hours per month			$23,408				$81,930			(Notes: staffing costs include cross-shifts)
S1.2.1.2		Office/Camp manager		6	months	360	site working hours per month			$22,267				$133,604			
S1.2.1.3		Administrative Assistants/HR/Accounting		4	months	360	site working hours per month			$16,600				$58,098			
S1.2.1.4		Water Treatment Staff Operator		5	months	360	site working hours per month			$18,912				$94,560			
S1.2.1.5		Environmental/Safety Manager		4	months	360	site working hours per month			$22,267				$89,069			
S1.2.1.6		Environmental Technicians		7	months	360	site working hours per month			$16,046				$104,301			
S1.2.1.7		Mechanic		2	months	360	site working hours per month			$17,769				$35,538			
S1.2.1.8		Tradesmen		2	months	360	site working hours per month			$18,912				$42,552			
S1.2.1.9		General Labour/helpers		10	months	360	site working hours per month			$12,793				$121,536			
S1.2.2	*Field support Vehicles*														$131,186		
S1.2.2.1		Pick-up trucks (3 required)		12	months					$4,446				$53,350			
S1.2.2.2		Mechanic service vehicle		2	months					$11,803				$23,607			
S1.2.2.3		Emergency transport vehicle		12	months					$4,519				$54,229			
S1.2.3	*Field Support Equipment/Supplies*														$232,440		
S1.2.3.1		Office supplies		12	months					$1,000				$12,000			
S1.2.3.2		Communications		12	months					$1,000				$12,000			
S1.2.3.3		Misc. supplies		12	months					$500				$6,000			
S1.2.3		Camp Operation		1,995	man-days					$80				$159,600			
S1.2.3		Power, heat, fuel		12	months					$3,570				$42,840			
S1.3	**Site Maintenance**														$359,187	**$359,187**	
S1.3.1		Equipment Maintenance		1	ls		$20,000							$20,000			Labour included in field support staff
S1.3.2		Road Maintenance	Grader (assume 80 hrs/month)	1	ls	960		$44	$106	$150	$0	$42,612	$101,443	$144,056			Operator cost included in Field support staff
S1.3.3		Earthwork Repair allowance (assume 40 hrs per month)		1	ls	480	$0	$132	$344	$344			$165,132	$165,132			Task Code C.2.14
S1.3.4		Sundry equipment/consumables allowance		1	ls					$30,000				$30,000			Allowance for pump maintenance/fuel etc.
S1.4	**Transportation Costs**														$151,714	**$151,714**	
S1.4.1		Barge Operations		5	months					$10,832				$54,162			
S1.4.2		Staffing Bus trips during barge operation period		20	ea					$678				$13,552			One per week
S1.4.3		Air transport and airstrip operations		8	flights					$3,000				$24,000			One flight per week
S1.4.4		Helicopter transport		20	flights					$3,000				$60,000			One flight per week
TOTAL														**$2,342,683**	**$2,342,683**	**$2,342,683**	

Phase 2 - Annual Care & Maintenance Costs During Active Closure

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S2.1	**Water Treatment**															**$706,967**	
S2.1.1	*Active Treatment System*														$706,967		
S2.1.1.1		Operation and maintenance		1	yr						$511,807	$65,194	$38,831	$615,832			See details on 'Water Treatment' worksheet.
S2.1.1.2		Capital Replacement	annual allowance	1	yr						$86,932	$4,203	$0	$91,135			See details on 'Water Treatment' worksheet.
S2.2	**Onsite Management**															**$310,135**	
S2.2.1	*Field Support Staff*														$247,815		
S2.2.1.1		Water Treatment Staff Operator		5	months	360	site working hours per month			$18,912				$94,560			
S2.2.1.2		Environmental/Safety Manager		4	months	360	site working hours per month			$22,267				$89,069			
S2.2.1.3		Environmental Technicians		4	months	360	site working hours per month			$16,046				$64,185			
S2.2.2	*Field support Vehicles*														$31,121		
S2.2.2.1		Pick-up trucks (1 required)		7	months					$4,446				$31,121			
S2.2.3	*Field Support Equipment/Supplies*														$31,200		
S2.2.3.1		Camp Operation		390	man-days					$80				$31,200			
S2.3	**Site Maintenance**														$30,000	**$30,000**	
S2.3.1		Pumping equipment/consumables allowance		1	ls					$30,000				$30,000			Allowance for pump maintenance/fuel etc.
S2.4	**Transportation Costs**														$0	**$0**	
S2.4.1		Barge Operations		0	months					$10,832				$0			Included in construction indirect costs
TOTAL														**$1,047,102**	**$1,047,102**	**$1,047,102**	

Phase 3 - Annual Care & Maintenance Costs - Post Closure I

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours or Cost Code Ref.	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S3.1	**Water Treatment**															**$741,982**	
S3.1.1	*Active Treatment System*														$706,967		
S3.1.1.1		Operation and maintenance		1	yr						$511,807	$65,194	$38,831	$615,832			See details on 'Water Treatment' worksheet.
S3.1.1.2		Capital Replacement	annual allowance	1	yr						$86,932	$4,203	$0	$91,135			See details on 'Water Treatment' worksheet.
S3.1.1	*Passive Treatment System*														$35,015		
S3.1.1.1		Operation and maintenance		1	yr						$10,000	$11,986	$13,029	$35,015			See details on 'Water Treatment' worksheet.
S3.2	**Onsite Management**															**$606,140**	
S3.2.1	*Field Support Staff*														$377,100		
S3.2.1.1		Mine Manager		3	months	360	site working hours per month			$23,408				$58,521			(Notes: staffing costs include cross-shifts)
S3.2.1.2		Office/Camp manager		3	months	360	site working hours per month			$22,267				$66,802			
S3.2.1.3		Administrative Assistants/HR/Accounting		1	months	360	site working hours per month			$16,600				$20,749			
S3.2.1.4		Water Treatment Staff Operator		2	months	360	site working hours per month			$18,912				$33,096			
S3.2.1.5		Environmental/Safety Manager		3	months	360	site working hours per month			$22,267				$55,668			
S3.2.1.6		Environmental Technicians		3	months	360	site working hours per month			$16,046				$40,116			
S3.2.1.7		Mechanic		2	months	360	site working hours per month			$17,769				$35,538			
S3.2.1.8		Tradesmen		2	months	360	site working hours per month			$18,912				$37,824			
S3.2.1.9		General Labour/helpers		2	months	360	site working hours per month			$12,793				$28,785			
S3.2.2	*Field support Vehicles*														$112,890		
S3.2.2.1		Pick-up trucks (3 required)		15	months					$4,446				$66,688			
S3.2.2.2		Mechanic service vehicle		2	months					$11,803				$23,607			
S3.2.2.3		Emergency transport vehicle		5	months					$4,519				$22,596			
S3.2.3	*Field Support Equipment/Supplies*														$116,150		
S3.2.3.1		Office supplies		5	months					$1,000				$5,000			
S3.2.3.2		Communications		5	months					$1,000				$5,000			
S3.2.3.3		Misc. supplies		5	months					$500				$2,500			
S3.2.3		Camp Operation		1,073	man-days					$80				$85,800			
S3.2.3		Power, heat, fuel		5	months					$3,570				$17,850			
S3.3	**Reclamation Maintenance**															**$366,811**	
S3.3.1	*Cover repairs*														$320,441		
S3.3.1.1		Cover Repair allowance	Load, haul, dump, spread (spoil lo	2%	of total	1,336,364	m3 of cover required replacement at:			$5.31	/m3			$142,044			Assumes 10% of area reseeded over 5 yr PC1 perio
S3.3.1.2			Seed/Fertilize: broadcast seeding	67	ha					$2,669.89	/ha			$178,397			
S3.3.2.1	*Revegetation*														$46,370		
S3.3.2.2		Reseeding allowance	Seed/Fertilize: broadcast seeding	5%	of total	347	ha requires reseeding at			$2,669.89	/ha			$46,370			Assumes 25% of area reseeded over 5 yr PC1 perio
S3.3	**Site Maintenance**														$99,414	**$99,414**	
S3.3.1		Equipment Maintenance		1	ls		$20,000							$20,000			Labour included in field support staff
S3.3.2		Road Maintenance	Grader (assume 40 hrs/month)	1	ls	200		$44	$106	$150	$0	$8,878	$21,134	$30,012			Operator cost included in Field support staff
S3.3.3		Earthwork Repair allowance (assume 20 hrs per month)		1	ls	100	$0	$132	$344	$344			$34,402	$34,402			Task Code C.2.14
S3.3.4		Sundry equipment/consumables allowance		1	ls					$15,000				$15,000			Allowance for pump maintenance/fuel etc.
S3.4	**Transportation Costs**														$64,628	**$64,628**	
S3.4.1		Barge Operations		3	months					$10,832				$32,497			
S3.4.2		Staffing Bus trips during barge operation period		12	ea					$678				$8,131			One per week
S3.4.3		Air transport and airstrip operations		8	flights					$3,000				$24,000			One flights per week
TOTAL														**$1,878,975**	**$1,878,975**	**$1,878,975**	

Phase 4 - Annual Care & Maintenance Costs - Post Closure II

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours or Cost Code Ref.	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S4.1	**Water Treatment**															**$126,150**	
S4.1.1	*Active Treatment System*														$91,135		
S4.1.1.2		Capital Replacement	annual allowance	1	yr						$86,932	$4,203	$0	$91,135			See details on 'Water Treatment' worksheet.
S4.1.1	*Passive Treatment System*														$35,015		
S4.1.1.1		Operation and maintenance		1	yr						$10,000	$11,986	$13,029	$35,015			See details on 'Water Treatment' worksheet.
S4.2	**Onsite Management**															**$79,830**	
S4.2.1	*Field Support Staff*														$44,847		
S4.2.1.1		Project Manager		0.5	months	360	site working hours per month			$23,408				$11,704			(Notes: staffing costs include cross-shifts)
S4.2.1.2		Passive Treatment specialist		0.5	months	360	site working hours per month			$42,120				$21,060			
S4.2.1.3		Mechanic		0.5	months	360	site working hours per month			$17,769				$8,885			
S4.2.1.4		General Labour/helpers		0.3	months	360	site working hours per month			$12,793				$3,198			
S4.2.2	*Field support Vehicles*														$19,313		
S4.2.2.1		Pick-up trucks (2 required)		2.0	months					$4,446				$8,892			
S4.2.2.2		Mechanic service vehicle		0.5	months					$11,803				$5,902			
S4.2.2.3		Emergency transport vehicle		1.0	months					$4,519				$4,519			
S4.2.3	*Field Support Equipment/Supplies*														$15,670		
S4.2.3.1		Office supplies		1	months					$1,000				$1,000			
S4.2.3.2		Communications		1	months					$1,000				$1,000			
S4.2.3.3		Misc. supplies		1	months					$500				$500			
S4.2.3		Camp Operation		120	man-days					$80				$9,600			
S4.2.3		Power, heat, fuel		1	months					$3,570				$3,570			
S4.3	**Reclamation Maintenance**															**$0**	
S4.3.1	*Cover repairs*														$0		
S4.3.2	*Revegetation*														$0		
S4.3	**Site Maintenance**														$29,882	**$29,882**	
S4.3.1		Equipment Maintenance (parts, supplies)		1	ls		$10,000							$10,000			Labour included in field support staff
S4.3.2		Road Maintenance	Grader (assume 20 hrs/month)	1	ls	20		$44	$106	$150	$0	$888	$2,113	$3,001			Operator cost included in Field support staff
S4.3.3		Earthwork Repair allowance (assume 20 hrs per month)		1	ls	20	$0	$132	$344	$344			$6,880	$6,880			Task Code C.2.14
S4.3.4		Sundry equipment/consumables allowance		1	ls					$10,000				$10,000			Allowance for pump maintenance/fuel etc.
S4.4	**Transportation Costs**														$20,832	**$20,832**	
S4.4.1		Barge Operations		1	months					$10,832				$10,832			
S4.4.2		Staffing Bus trips during barge operation period		0	ea					$678				$0			Pick-up trucks used (time included in labour rate)
S4.4.3		Barge mob/demob		1	LS					$10,000				$10,000			
TOTAL														**$256,693**	**$256,693**	**$256,693**	

Phase 5 - Perpetual Care & Maintenance Event Year Annual Costs

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours or Cost Code Ref.	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S5.1	**Water Treatment**															**$35,015**	
S5.1.1	*Passive Treatment System*														$35,015		
S5.1.1.1		Operation and maintenance		1	yr						$10,000	$11,986	$13,029	$35,015			See details on 'Water Treatment' worksheet.
S5.2	**Onsite Management**															**$178,636**	
S5.2.1	*Field Support Staff*														$129,326		
S5.2.1.1		Project Manager		1.0	months	360	site working hours per month			$23,408				$23,408			(Notes: staffing costs include cross-shifts)
S5.2.1.2		Passive Treatment specialist		1.0	months	360	site working hours per month			$42,120				$42,120			
S5.2.1.3		Site Engineer		1.0	months	360	site working hours per month			$42,120				$42,120			
S5.2.1.4		HD mechanic		0.5	months	360	site working hours per month			$17,769				$8,885			
S5.2.1.5		General Labour/helpers		1.0	months	360	site working hours per month			$12,793				$12,793			
S5.2.2	*Field support Vehicles*														$19,239		
S5.2.2.1		Pick-up trucks (3 required)		3	months					$4,446				$13,338			
S5.2.2.2		Mechanic service vehicle		1	months					$11,803				$5,902			
S5.2.3	*Field Support Equipment/Supplies*														$30,070		
S5.2.3.1		Office supplies		1	months					$1,000				$1,000			
S5.2.3.2		Communications		1	months					$1,000				$1,000			
S5.2.3.3		Misc. supplies		1	months					$500				$500			
S5.2.3.4		Camp Operation		300	man-days					$80				$24,000			
S5.2.3.5		Power, heat, fuel		1	months					$3,570				$3,570			
S5.3	**Reclamation Maintenance**															**$0**	
S5.4	**Site Maintenance**														$108,253	**$108,253**	
S5.4.1		Grader Maintenance	Grader (assume 2 weeks, 10hrs/d	1	ls	140		$44	$106	$150	$0	$6,214	$14,794	$21,008			Operator cost included in Field support staff
S5.4.2		Earthwork Repair allowance (assume 3 weeks, 10hrs/day)		1	ls	210	$0	$132	$344	$344			$72,245	$72,245			Task Code C.2.14
S5.4.3		Big Creek Bridge - Capital replacement allowance		1	ls					$15,000				$15,000			Allowance for supplies, pumping, etc.
S5.4.4		Sundry allowance		1	ls					$2,500				$2,500			Allowance for supplies, pumping, etc.
S5.5	**Transportation Costs**														$69,943	**$69,943**	
S5.5.1		Barge Operations		1	months					$10,832				$10,832			
S5.5.2		Barge mobilization/demob and set up		1	ls					$10,000				$10,000			assumed
S5.5.3		Camp Mob/demob		1	ls					$10,000				$10,000			assumed
S5.5.4		Equipment mobilization and demobilization		2	ls					$19,555				$39,110.91			See details on 'mob-demob 'worksheet
TOTAL														**$394,347**	**$391,847**	**$391,847**	

Worksheet 9 - Peak Liability Estimate - Implementation Costs

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Peak Liability Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
DIRECT COSTS																					
A1	**Waste Dumps**																			**$6,893,723**	
A1.1	*Main Waste Dump (inc. MWDE)*																		$121,985		
A1.1.1.1		Regrade	Flat areas: Re-grade to form tertiary drainage catchments	0	hrs	C.2.13	0	$44.39	$0	$194.68	$0	$67.01	0	$0	$0.00	$0	$306.07	$0			
A1.1.1.2			Slopes: Re-grade to 3H:1V	0	hrs	C.2.13	0	$44.39	$0	$194.68	$0	$67.01	0	$0	$0.00	$0	$306.07	$0			
A1.1.2.1		Cover	Flat area cover material: Load, haul, dump spread	0	Cm3	R.050	0.0	$0.82	$0	$2.67	$0	$0.88	0	$0	$0.00	$0	$4.37	$0			
A1.1.2.2			Slope area cover material: Load, haul, dump along crest	0	Cm3	R.051	0.0	$0.72	$0	$2.37	$0	$0.80	0	$0	$0.00	$0	$3.89	$0			
A1.1.2.3			Slope area cover material: Spread down slope	0	hrs	C.2.13	0.0	$44.39	$0	$194.68	$0	$67.01	0	$0	$0.00	$0	$306.07	$0			
A1.1.3.1		Revegetate	Seed/Fertilize: tractor application	0	ha	C.6.01	0.0	$1,029.50	$0	$401.72	$0	$232.94	0	$0	$1,005.72	$0	$2,670	$0			
A1.1.3.2			Tree seedling application (1,000 stems/ha)	52	ha	C.6.06	2,668.2	$1,832.33	$95,397	$25.98	$1,353	$15.00	813	$781	$469.71	$24,455	$2,343	$121,985			
A1.2	*Main Waste Dump Wrap*																		$350,260		
A1.2.1.1		Regrade	Flat areas: Re-grade to form tertiary drainage catchments	6	hrs	C.2.13	6	$44.39	$278	$194.68	$1,218	$67.01	437	$419	$0.00	$0	$306.07	$1,914			
A1.2.1.2			Slopes: Re-grade to 3H:1V	255	hrs	C.2.13	255	$44.39	$11,334	$194.68	$49,709	$67.01	17,822	$17,110	$0.00	$0	$306.07	$78,152			
A1.2.2.1		Cover	Flat area cover material: Load, haul, dump spread	8,223	Cm3	R.050	153.4	$0.82	$6,757	$2.67	$21,957	$0.88	7,527	$7,226	$0.00	$0	$4.37	$35,939			
A1.2.2.2			Slope area cover material: Load, haul, dump along crest	42,447	Cm3	R.051	692.9	$0.72	$30,484	$2.37	$100,640	$0.80	35,215	$33,806	$0.00	$0	$3.89	$164,930			
A1.2.2.3			Slope area cover material: Spread down slope	61	hrs	C.2.13	60.5	$44.39	$2,686	$194.68	$11,781	$67.01	4,224	$4,055	$0.00	$0	$306.07	$18,523			
A1.2.3.1		Revegetate	Seed/Fertilize: tractor application	10	ha	C.6.01	335.7	$1,029.50	$10,433	$401.72	$4,071	$232.94	2,459	$2,361	$1,005.72	$10,192	$2,670	$27,057			
A1.2.3.2			Tree seedling application (1,000 stems/ha)	10	ha	C.6.06	519.4	$1,832.33	$18,569	$25.98	$263	$15.00	158	$152	$469.71	$4,760	$2,343	$23,744			
A1.3	*Southwest Dump (excluding high grade waste area)*																		$147,053		
A1.3.1		Re-grade	Slopes: Re-grade remaining to 3H:1V	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A1.3.2.1		Cover	Load, haul, dump spread	0	m3	R.059	0.0	$0.77	$0.00	$2.49	$0	$0.82	0	$0.00	$0.00	$0.00	$4.08	$0			
A1.3.3.1		Revegetate	Seed/Fertilize: tractor application	0.00	ha	C.6.01	0.0	$1,029.50	$0.00	$401.72	$0	$232.94	0	$0.00	$1,005.72	$0.00	$2,669.89	$0			
A1.3.3.2			Tree seedling application (1,000 stems/ha)	62.76	ha	C.6.06	3,216.6	$1,832.33	$115,000.98	$25.98	$1,631	$15.00	981	$941.43	$469.71	$29,479.97	$2,343.03	$147,053			
A1.4	*Southwest Dump High-Grade Waste*																		$249,632		
A1.4.1		Relocate High-grade waste	Load, haul, dump, spread in pit (below long-term water elevation)	45,711	Bm3	R.069	688.0	$0.66	$30,283.89	$2.19	$100,276	$0.73	34,965	$33,566.76	$0.00	$0.00	$3.59	$164,127			
A1.4.2		Regrade	Flat areas: Re-grade to form tertiary drainage catchments	31	hrs	C.2.13	31.1	$44.39	$1,378.38	$194.68	$6,045	$67.01	2,168	$2,080.81	$0.00	$0.00	$306.07	$9,505			
A1.4.3		Cover	Flat area cover material: Load, haul, dump spread	15,583	Cm3	R.060	257.8	$0.73	$11,362.15	$2.37	$36,935	$0.78	12,585	$12,081.20	$0.00	$0.00	$3.87	$60,378			
A1.4.4.1		Revegetate	Seed/Fertilize: broadcast seeding	3.12	ha	C.6.01	103.2	$1,029.50	$3,208.44	$401.72	$1,252	$232.94	756	$725.97	$1,005.72	$3,134.33	$2,669.89	$8,321			
A1.4.4.2			Tree seedling application (1,000 stems/ha)	3.12	ha	C.6.06	159.7	$1,832.33	$5,710.47	$25.98	$81	$15.00	49	$46.75	$469.71	$1,463.85	$2,343.03	$7,302			
A1.5	*Main Pit SAT*																		$4,009,367		
A1.5.1		Relocate SAT below water	Load, haul, dump, spread in pit	1,040,000	Cm3	R.012	17,781.5	$0.75	$785,076.51	$1.86	$1,935,773	$0.51	555,290	$533,078.26	$0.00	$0.00	$3.13	$3,253,928			
A1.5.2		Backfill SAT Excavation	Load, haul, dump WR to form buttress below MPD	245,000	Cm3	R.013	3,226.9	$0.58	$142,343.39	$1.89	$462,905	$0.61	156,449	$150,191.13	$0.00	$0.00	$3.08	$755,439			Cover and revegetation of backfill included in MPD (A1.6)
A1.6	*Main Pit Dump (Incl. SWB, In-Pit Dumps)*																		$686,521		
A1.6.1.1		Regrade	Flat areas: Re-grade to form tertiary drainage catchments	65	hrs	C.2.13	65.1	$44.39	$2,891.84	$194.68	$12,683	$67.01	4,547	$4,365.54	$0.00	$0.00	$306.07	$19,941			
A1.6.1.2			Slopes: Re-grade to 3H:1V	207	hrs	C.2.13	206.7	$44.39	$9,174.86	$194.68	$40,240	$67.01	14,428	$13,850.42	$0.00	$0.00	$306.07	$63,265			
A1.6.2.1		Cover	Flat area cover material: Load, haul, dump spread	45,270	m3	R.053	732.0	$0.71	$32,263.18	$2.32	$104,877	$0.76	35,734	$34,304.93	$0.00	$0.00	$3.79	$171,445			
A1.6.2.2			Slope area cover material: Load, haul, dump along crest	84,053	m3	R.054	1,165.0	$0.61	$51,284.32	$2.02	$169,813	$0.68	59,212	$56,843.70	$0.00	$0.00	$3.31	$277,941			
A1.6.2.3			Slope area cover material: Spread down slope	79	hrs	C.2.13	79.3	$44.39	$3,520.01	$194.68	$15,438	$67.01	5,535	$5,313.82	$0.00	$0.00	$306.07	$24,272			
A1.6.3.1		Revegetate	Seed/Fertilize: broadcast seeding	25.9	ha	C.6.01	856.8	$1,029.50	$26,627.72	$401.72	$10,390	$232.94	6,276	$6,025.02	$1,005.72	$26,012.60	$2,669.89	$69,056			
A1.6.3.2			Tree seedling application (1,000 stems/ha)	25.9	ha	C.6.06	1,325.6	$1,832.33	$47,392.66	$25.98	$672	$15.00	404	$387.97	$469.71	$12,148.89	$2,343.03	$60,602			
A1.6.0.0	*Mill Valley Fill Extension (Stage 1 and 2)*																		$40,934		
A1.6.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A1.6.1.2			Slopes: Re-grade to 3H:1V	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A1.6.2.1		Cover	Flat area cover material: Load, haul, dump spread	0	m3	R.056	0.0	$0.90	$0.00	$2.91	$0	$0.96	0	$0.00	$0.00	$0.00	$4.77	$0			
A1.6.2.2			Slope area cover material: Load, haul, dump along crest	0	m3	R.057	0.0	$0.78	$0.00	$2.59	$0	$0.87	0	$0.00	$0.00	$0.00	$4.24	$0			
A1.6.2.3			Slope area cover material: Spread down slope	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A1.6.3.1		Revegetate	Seed/Fertilize: broadcast seeding	0.00	ha	C.6.01	0.0	$1,029.50	$0.00	$401.72	$0	$232.94	0	$0.00	$1,005.72	$0.00	$2,669.89	$0			
A1.6.3.2			Tree seedling application (1,000 stems/ha)	17.47	ha	C.6.06	895.4	$1,832.33	$32,011.85	$25.98	$454	$15.00	273	$262.06	$469.71	$8,206.09	$2,343.03	$40,934			
A1.7.0.0	*Ridgetop Waste Dump*																		$1,287,970		
A1.7.1.1		Regrade	Flat areas: Re-grade to form tertiary drainage catchments	88	hrs	C.2.13	88	$44.39	$3,903	$194.68	$17,119	$67.01	6,138	$5,892	$0.00	$0	$306.07	$26,915			
A1.7.1.2			Slopes: Re-grade to 3H:1V	383	hrs	C.2.13	383	$44.39	$17,021	$194.68	$74,651	$67.01	26,765	$25,695	$0.00	$0	$306.07	$117,367			
A1.7.2.1		Cover	Flat area cover material: Load, haul, dump spread	22,063	Cm3	R.063	590.5	$1.18	$26,007	$3.83	$84,514	$1.26	28,971	$27,812	$0.00	$0	$6.27	$138,332			
A1.7.2.2			Slope area cover material: Load, haul, dump along crest	142,998	Cm3	R.064	3,348.7	$1.03	$147,324	$3.40	$486,381	$1.14	170,188	$163,381	$0.00	$0	$5.57	$797,086			
A1.7.2.3			Slope area cover material: Spread down slope	140	hrs	C.2.13	139.8	$44.39	$6,205	$194.68	$27,213	$67.01	9,757	$9,367	$0.00	$0	$306.07	$42,784			
A1.7.3.1		Revegetate	Seed/Fertilize: tractor application	33	ha	C.6.01	1,093.6	$1,029.50	$33,986	$401.72	$13,262	$232.94	8,010	$7,690	$1,005.72	$33,201	$2,670	$88,139			
A1.7.3.2			Tree seedling application (1,000 stems/ha)	33	ha	C.6.06	1,691.9	$1,832.33	$60,489	$25.98	$858	$15.00	516	$495	$469.71	$15,506	$2,343	$77,348			
A2	**Overburden Dumps**																			**$395,276**	
A2.1	*Area 118 Pit Backfill Dump*																		$16,268		
A2.1.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A2.1.1.2			Slopes: Regrade to 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A2.1.2.1		Revegetate	Seed/Fertilize: broadcast seeding	0	ha	C.6.01	0.0	$1,029.50	$0.00	$401.72	$0	$232.94	0	$0.00	$1,005.72	$0.00	$2,669.89	$0			
A2.1.2.2			Tree seedling application (1,000 stems/ha)	7	ha	C.6.06	355.8	$1,832.33	$12,722.54	$25.98	$180	$15.00	108	$104.15	$469.71	$3,261.36	$2,343.03	$16,268			
A2.2	*Ice-rich Overburden Dump*																		$13,827		
A2.2.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A2.2.1.2			Berm: Regrade to 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			WR Berm area included in Southwest Dump
A2.2.3.1		Revegetate	Seed/Fertilize: broadcast seeding	2.76	ha	C.6.01	91.4	$1,029.50	$2,839.68	$401.72	$1,108	$232.94	669	$642.53	$1,005.72	$2,774.08	$2,669.89	$7,364			
A2.2.3.2			Tree seedling application (1,000 stems/ha)	2.76	ha	C.6.06	141.4	$1,832.33	$5,054.12	$25.98	$72	$15.00	43	$41.37	$469.71	$1,295.60	$2,343.03	$6,463			
A2.3	*Reclamation Overburden Dump*																		$330,992		
A2.3.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	594	hrs	C.2.13	593.6	$44.39	$26,347.50	$194.68	$115,557	$67.01	41,432	$39,774.35	$0.00	$0.00	$306.07	$181,679			
A2.3.1.2			Slopes: Regrade to 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			Assumes borrow removed in such a manner that significant resloping is not required.
A2.3.2.1		Revegetate	Seed/Fertilize: broadcast seeding	29.79	ha	C.6.01	986.7	$1,029.50	$30,664.45	$401.72	$11,966	$232.94	7,228	$6,938.40	$1,005.72	$29,956.07	$2,669.89	$79,525			
A2.3.2.2			Tree seedling application (1,000 stems/ha)	29.79	ha	C.6.06	1,526.5	$1,832.33	$54,577.32	$25.98	$774	$15.00	465	$446.79	$469.71	$13,990.64	$2,343.03	$69,789			
A2.4	*Ridgetop South Backfill Dump*																		$34,189		
A2.4.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	17	hrs	C.2.13	16.9	$44.39	$751.05	$194.68	$3,294	$67.01	1,181	$1,133.78	$0.00	$0.00	$306.07	$5,179			
A2.4.1.2			Slopes: Regrade to 3H:1V or flatter	32	hrs	C.2.13	32.2	$44.39	$1,430.97	$194.68	$6,276	$67.01	2,250	$2,160.20	$0.00	$0.00	$306.07	$9,867			
A2.4.2.1		Revegetate	Seed/Fertilize: broadcast seeding	3.82	ha	C.6.01	126.5	$1,029.50	$3,931.30	$401.72	$1,534	$232.94	927	$889.53	$1,005.72	$3,840.48	$2,669.89	$10,195			
A2.4.2.2			Tree seedling application (1,000 stems/ha)	3.82	ha	C.6.06	195.7	$1,832.33	$6,997.02	$25.98	$99	$15.00	60	$57.28	$469.71	$1,793.65	$2,343.03	$8,947			
A3	**Ore Stockpiles**																			**$3,331,999**	
A3.1	*High and Medium Grade Sulphide Ore Stockpile (South of Mill)*																		$3,176,985		(North, West, East, and South Stockpiles)
A3.1.1.1		Relocate to Pit	Ore: Load, haul, dump, spread into Main Pit	884,785	m3	R.044	10,408.2	$0.52	$458,552.27	$1.72	$1,524,620	$0.57	529,049	$507,886.56	$0.00	$0.00	$2.82	$2,491,059			
A3.1.1.2			Ore Pads (over-excavate): Load, haul, dump spread into Main Pit	71,007	m3	R.044	835.3	$0.52	$36,800.28	$1.72	$122,356	$0.57	42,458	$40,759.51	$0.00	$0.00	$2.82	$199,915			
A3.1.1.3			Complete confirmation testing	1,421	ea	C.3.02	236.8	$7.43	$10,556.37	$1.33	$1,883	$0.00	0	$0.00	$13.26	$18,842.46	$22.01	$31,282			
A3.1.2.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	289	hrs	C.2.13	289.4	$44.39	$12,843.71	$194.68	$56,331	$67.01	20,197	$19,388.95	$0.00	$0.00	$306.07	$88,564			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Peak Liability Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A3.1.3.1		Cover	Cover material: Load, haul, dump spread (0.5m)	69,653	m3	R.055	1,264.9	$0.80	$55,711.18	$2.60	$181,043	$0.86	62,060	$59,577.97	$0.00	$0.00	$4.25	$296,332			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Peak Liability Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A3.1.4.1		Revegetate	Seed/Fertilize: broadcast seeding	13.93	ha	C.6.01	461.5	$1,029.50	$14,341.67	$401.72	$5,596	$232.94	3,380	$3,245.07	$1,005.72	$14,010.36	$2,669.89	$37,193			
A3.1.4.2			Tree seedling application (1,000 stems/ha)	13.93	ha	C.6.06	713.9	$1,832.33	$25,525.64	$25.98	$362	$15.00	218	$208.96	$469.71	$6,543.38	$2,343.03	$32,640			
A3.2	*Low Grade Sulphide Ore Stockpile (Main Dump Area)*																		$155,014		
A3.2.1.1		Relocate to Pit	Ore: Load, haul, dump, spread into Main Pit	15,490	m3	R.036	195.9	$0.56	$8,624.35	$1.84	$28,557	$0.62	9,958	$9,559.25	$0.00	$0.00	$3.02	$46,741			
A3.2.1.2			Ore Pad (over-excavate): Load, haul, dump spread into Main Pit	31,316	m3	R.036	396.1	$0.56	$17,435.36	$1.84	$57,732	$0.62	20,131	$19,325.41	$0.00	$0.00	$3.02	$94,493			
A3.2.1.3			Complete confirmation testing	626	ea	C.3.02	104.3	$7.43	$4,650.45	$1.33	$829	$0.00	0	$0.00	$13.26	$8,300.76	$22.01	$13,781			
A3.2.2.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			Included in Main Waste Dump Closure
A3.2.3.1		Cover	Cover material: Load, haul, dump spread	0	m3	R.050	0.0	$0.82	$0.00	$2.67	$0	$0.88	0	$0.00	$0.00	$0.00	$4.37	$0			Included in Main Waste Dump Closure
A3.2.4.1		Revegetate	Seed/Fertilize: broadcast seeding	0	ha	C.6.01	0.0	$1,029.50	$0.00	$401.72	$0	$232.94	0	$0.00	$1,005.72	$0.00	$2,669.89	$0			Included in Main Waste Dump Closure
A3.2.4.2			Tree seedling application (1,000 stems/ha)	0	ha	C.6.06	0.0	$1,832.33	$0.00	$25.98	$0	$15.00	0	$0.00	$469.71	$0.00	$2,343.03	$0			Included in Main Waste Dump Closure
A4	**Open Pits**																			**$620,582**	
A4.1	*Main Pit*																		$150,984		
A4.1.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	7,500	m2	C.2.03	28.5	$0.17	$1,265.55	$0.49	$3,658	$0.13	1,048	$1,005.87	$0.00	$0.00	$0.79	$5,930			
A4.1.1.2			Safety Berm: Construct around highwall perimeter	4,860	m3	C.2.02	45.8	$0.42	$2,035.14	$0.89	$4,333	$0.25	1,267	$1,215.92	$0.00	$0.00	$1.56	$7,584			
A4.1.1.3			Place large boulders across any pit access points	52	m3	R.001	1.6	$1.39	$72.20	$3.03	$157	$0.90	49	$46.76	$0.00	$0.00	$5.31	$276			
A4.1.1.4			Allowance: Install warning signs around pit perimeter at key locations	6	ea	C.5.13	5.6	$40.27	$241.63	$15.90	$95	$5.57	35	$33.39	$199.26	$1,195.57	$261.00	$1,566			
A4.1.2.1		Exposed tailings cover	Trafficability Layer: Load, haul, dump spread	17,750	m3	R.014	554.4	$1.38	$24,425.58	$2.86	$50,708	$0.82	15,091	$14,487.41	$0.00	$0.00	$5.05	$89,621			
A4.1.2.2			Overburden Layer: Load, haul, dump spread	8,875	m3	R.052	158.4	$0.79	$6,976.73	$2.55	$22,672	$0.84	7,772	$7,460.97	$0.00	$0.00	$4.18	$37,110			
A4.1.3.1		Revegetate	Seed/Fertilize: broadcast seeding	2	ha	C.6.01	58.8	$1,029.50	$1,827.37	$401.72	$713	$232.94	431	$413.48	$1,005.72	$1,785.15	$2,669.89	$4,739			
A4.1.3.2			Tree seedling application (1,000 stems/ha)	2	ha	C.6.06	91.0	$1,832.33	$3,252.39	$25.98	$46	$15.00	28	$26.63	$469.71	$833.74	$2,343.03	$4,159			
A4.2	*Area 2 Pit*																		$10,001		
A4.2.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	0	m2	C.2.03	0.0	$0.17	$0.00	$0.49	$0	$0.13	0	$0.00	$0.00	$0.00	$0.79	$0			
A4.2.1.2			Safety Berm: Construct around highwall perimeter	5,317	m3	C.2.02	50.2	$0.42	$2,226.54	$0.89	$4,740	$0.25	1,386	$1,330.27	$0.00	$0.00	$1.56	$8,297			
A4.2.1.3			Place large boulders across any pit access points	26	m3	R.001	0.8	$1.39	$36.10	$3.03	$79	$0.90	24	$23.38	$0.00	$0.00	$5.31	$138			
A4.2.1.4			Allowance: Install warning signs around pit perimeter at key locations	6	ea	C.5.13	5.6	$40.27	$241.63	$15.90	$95	$5.57	35	$33.39	$199.26	$1,195.57	$261.00	$1,566			
A4.3	*Area 118 Pit*																		$0		
A4.3.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	0	m2	C.2.03	0.0	$0.17	$0.00	$0.49	$0	$0.13	0	$0.00	$0.00	$0.00	$0.79	$0			Pit filled it by Area 118 Backfill Dump
A4.3.1.2			Safety Berm: Construct around highwall perimeter	0	m3	C.2.02	0.0	$0.42	$0.00	$0.89	$0	$0.25	0	$0.00	$0.00	$0.00	$1.56	$0			
A4.3.1.3			Place large boulders across any pit access points	0	m3	R.001	0.0	$1.39	$0.00	$3.03	$0	$0.90	0	$0.00	$0.00	$0.00	$5.31	$0			
A4.3.1.4			Allowance: Install warning signs around pit perimeter at key locations	0	ea	C.5.13	0.0	$40.27	$0.00	$15.90	$0	$5.57	0	$0.00	$199.26	$0.00	$261.00	$0			
A4.4	*Minto North Pit*																		$11,481		
A4.4.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	5,900	m2	C.2.03	22.4	$0.17	$995.56	$0.49	$2,878	$0.13	824	$791.29	$0.00	$0.00	$0.79	$4,665			
A4.4.1.2			Safety Berm: Construct around highwall perimeter	3,610	m	C.2.02	34.1	$0.42	$1,511.81	$0.89	$3,219	$0.25	941	$903.25	$0.00	$0.00	$1.56	$5,634			
A4.4.1.3			Place large boulders across any pit access points	26	m3	R.001	0.8	$1.39	$36.10	$3.03	$79	$0.90	24	$23.38	$0.00	$0.00	$5.31	$138			
A4.4.1.4			Allowance: Install warning signs around pit perimeter at key locations	4	ea	C.5.13	3.7	$40.27	$161.09	$15.90	$64	$5.57	23	$22.26	$199.26	$797.04	$261.00	$1,044			
A4.5	*Ridgetop North*																		$448,116		
A4.5.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	10,110	m2	C.2.03	38.4	$0.17	$1,705.95	$0.49	$4,931	$0.13	1,412	$1,355.92	$0.00	$0.00	$0.79	$7,993			
A4.5.1.2			Safety Berm: Construct around highwall perimeter	3,648	m	C.2.02	34.4	$0.42	$1,527.56	$0.89	$3,252	$0.25	951	$912.66	$0.00	$0.00	$1.56	$5,692			
A4.5.1.3			Place large boulders across any pit access points	26	m3	R.001	0.8	$1.39	$36.10	$3.03	$79	$0.90	24	$23.38	$0.00	$0.00	$5.31	$138			
A4.5.1.4			Allowance: Install warning signs around pit perimeter at key locations	6	ea	C.5.13	5.6	$40.27	$241.63	$15.90	$95	$5.57	35	$33.39	$199.26	$1,195.57	$261.00	$1,566			
A4.5.2.1		Exposed tailings cover	Trafficability Layer: Load, haul, dump spread	62,477	m3	R.068	816.5	$0.58	$35,941.63	$1.90	$119,010	$0.64	41,498	$39,837.81	$0.00	$0.00	$3.12	$194,789			
A4.5.2.2			Overburden Layer: Load, haul, dump spread	31,239	m3	R.062	882.0	$1.24	$38,844.69	$4.04	$126,233	$1.33	43,272	$41,540.81	$0.00	$0.00	$6.61	$206,618			
A4.5.3.1		Revegetate	Seed/Fertilize: broadcast seeding	6	ha	C.6.01	207.0	$1,029.50	$6,432.02	$401.72	$2,510	$232.94	1,516	$1,455.36	$1,005.72	$6,283.44	$2,669.89	$16,681			
A4.5.3.2			Tree seedling application (1,000 stems/ha)	6	ha	C.6.06	320.2	$1,832.33	$11,447.87	$25.98	$162	$15.00	98	$93.72	$469.71	$2,934.61	$2,343.03	$14,639			
A4.6	*Ridgetop South*																		$0		
A4.6.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	0	m2	C.2.03	0.0	$0.17	$0.00	$0.49	$0	$0.13	0	$0.00	$0.00	$0.00	$0.79	$0			Piit to be backfilled with overburden from Ridgetop North
A4.6.1.2			Safety Berm: Construct around highwall perimeter	0	m	C.2.02	0.0	$0.42	$0.00	$0.89	$0	$0.25	0	$0.00	$0.00	$0.00	$1.56	$0			
A4.6.1.3			Place large boulders across any pit access points	0	m3	R.001	0.0	$1.39	$0.00	$3.03	$0	$0.90	0	$0.00	$0.00	$0.00	$5.31	$0			
A4.6.1.4			Allowance: Install warning signs around pit perimeter at key locations	0	ea	C.5.13	0.0	$40.27	$0.00	$15.90	$0	$5.57	0	$0.00	$199.26	$0.00	$261.00	$0			
A5	**Underground Openings**																			**$81,637**	
A5.1	*Minto North Portal*																		$16,573		
A5.1.1.1		Seal portal	Load, Haul, Dump backfill plug	2,588	m3	R.001	81.5	$1.39	$3,592.52	$3.03	$7,832	$0.90	2,423	$2,326.51	$0.00	$0.00	$5.31	$13,751			Material sourced from nearby laydown area rockfill
A5.1.1.2			Doze backfill plug into portal with small dozer	16	hrs	C.2.19	0.0	$44.39	$723.90	$105.36	$1,718	$23.28	395	$379.66	$0.00	$0.00	$173.02	$2,822			
A5.2	*Minto North Portal Egress*																		$16,573		
A5.2.1.1		Seal portal	Load, Haul, Dump backfill plug	2,588	m3	R.001	81.5	$1.39	$3,592.52	$3.03	$7,832	$0.90	2,423	$2,326.51	$0.00	$0.00	$5.31	$13,751			Material sourced from nearby laydown area rockfill
A5.2.1.2			Doze backfill plug into portal with small dozer	16	hrs	C.2.19	0.0	$44.39	$723.90	$105.36	$1,718	$23.28	395	$379.66	$0.00	$0.00	$173.02	$2,822			
A5.3	*Minto South Portal*																		$16,573		
A5.3.1.1		Seal portal	Load, Haul, Dump backfill plug	2,588	m3	R.001	81.5	$1.39	$3,592.52	$3.03	$7,832	$0.90	2,423	$2,326.51	$0.00	$0.00	$5.31	$13,751			Material sourced from nearby laydown area rockfill
A5.3.1.2			Doze backfill plug into portal with small dozer	16	hrs	C.2.19	0.0	$44.39	$723.90	$105.36	$1,718	$23.28	395	$379.66	$0.00	$0.00	$173.02	$2,822			
A5.3.2.1		Revegetate	Included in Yards (WBS No. A11-10)																		
A5.4	*Area 118 Vent Raise*																		$10,691		
A5.4.1.1		Site Preparation	Removal of Heater/Fan included in Demolition																		
A5.4.1.2			Excavate and clean perimeter around shaft to suitable material.	4	ea	C.8.01	8.0	$84.00	$335.99	$13.66	$55	$6.83	28	$27.33	$0.00	$0.00	$104.49	$418			
A5.4.2.1		Concrete Cap installation	Structural steel - Supply and install I-Beam structural support	8	m	C.8.02	9.8	$50.19	$401.48	$15.95	$128	$6.05	50	$48.37	$16.09	$128.68	$88.27	$706			
A5.4.2.2			Formwork: Supply and install concrete ring-wall formwork	8	m2	C.8.03	51.3	$309.43	$2,475.43	$0.00	$0	$0.00	0	$0.00	$3.54	$28.32	$312.97	$2,504			
A5.4.2.3			Ringwall Rebar: supply and install	32	m	C.8.05	0.8	$1.27	$40.74	$0.00	$0	$0.00	0	$0.00	$3.37	$107.82	$4.64	$149			
A5.4.2.4			Concrete slab: supply and install	8.6	m3	C.8.06	61.1	$329.23	$2,844.57	$10.63	$92	$3.50	31	$30.21	$386.22	$3,336.96	$729.58	$6,304			Includes slab formwork, rebar, finishing.
A5.4.2.5			Vent Raise Pipe Supply and install	1	ea.	C.8.07	0.5	$17.77	$17.77	$0.00	$0	$0.00	0	$0.00	$460.62	$460.62	$478.39	$478			
A5.4.3.1		Backfill cap area	Backfill: Load, haul, dump, place over concrete cap	25	m3	R.001	0.8	$1.39	$34.71	$3.03	$76	$0.90	23	$22.48	$0.00	$0.00	$5.31	$133			
A5.5	*Minto East Vent Raise*																		$10,610		
A5.5.1.2		Preparation	Excavate and clean perimeter around shaft to suitable material.	4	m3	C.8.01	8.0	$84.00	$335.99	$13.66	$55	$6.83	28	$27.33	$0.00	$0.00	$104.49	$418			
A5.5.2.1		Concrete Cap installation	Structural steel - Supply and install I-Beam structural support	8	m	C.8.02	9.8	$50.19	$401.48	$15.95	$128	$6.05	50	$48.37	$16.09	$128.68	$88.27	$706			
A5.5.2.2			Formwork: Supply and install concrete ring-wall formwork	8	m2	C.8.03	51.3	$309.43	$2,475.43	$0.00	$0	$0.00	0	$0.00	$3.54	$28.32	$312.97	$2,504			
A5.5.2.3			Ringwall Rebar: supply and install	14	m	C.8.05	0.3	$1.27	$18.33	$0.00	$0	$0.00	0	$0.00	$3.37	$48.52	$4.64	$67			
A5.5.2.4			Concrete slab: supply and install	8.6	m3	C.8.06	61.1	$329.23	$2,844.57	$10.63	$92	$3.50	31	$30.21	$386.22	$3,336.96	$729.58	$6,304			Includes slab formwork, rebar, finishing.
A5.5.2.5			Vent Raise Pipe Supply and install	1	ea.	C.8.07	0.5	$17.77	$17.77	$0.00	$0	$0.00	0	$0.00	$460.62	$460.62	$478.39	$478			
A5.5.3.1		Backfill cap area	Backfill: Load, haul, dump, place over concrete cap	25	m3	R.001	0.8	$1.39	$34.71	$3.03	$76	$0.90	23	$22.48	$0.00	$0.00	$5.31	$133			
A5.6	*Minto East 2 Vent Raise*																		$10,617		
A5.6.1.1		Site Preparation	Removal of Heater/Fan included in Demolition																		
A5.6.1.2			Excavate and clean perimeter around shaft to suitable material.	4	m3	C.8.01	8.0	$84.00	$335.99	$13.66	$55	$6.83	28	$27.33	$0.00	$0.00	$104.49	$418			
A5.6.2.1		Concrete Cap installation	Structural steel - Supply and install I-Beam structural support	8	m	C.8.02	9.8	$50.19	$401.48	$15.95	$128	$6.05	50	$48.37	$16.09	$128.68	$88.27	$706			
A5.6.2.2			Formwork: Supply and install concrete ring-wall formwork	8	m2	C.8.03	51.3	$309.43	$2,475.43	$0.00	$0	$0.00	0	$0.00	$3.54	$28.32	$312.97	$2,504			
A5.6.2.3			Ringwall Rebar: supply and install	16	m	C.8.05	0.4	$1.27	$20.37	$0.00	$0	$0.00	0	$0.00	$3.37	$53.91	$4.64	$74			
A5.6.2.4			Concrete slab: supply and install	8.6	m3	C.8.06	61.1	$329.23	$2,844.57	$10.63	$92	$3.50	31	$30.21	$386.22	$3,336.96	$729.58	$6,304			Includes slab formwork, rebar, finishing.
A5.6.2.5			Vent Raise Pipe Supply and install	1	ea.	C.8.07	0.5	$17.77	$17.77	$0.00	$0	$0.00	0	$0.00	$460.62	$460.62	$478.39	$478			
A5.6.3.1		Backfill cap area	Backfill: Load, haul, dump, place over concrete cap	25	m3	R.001	0.8	$1.39	$34.71	$3.03	$76	$0.90	23	$22.48	$0.00	$0.00	$5.31	$133			
A6	**External Tailings Facilities**																			**$57,291**	
A6.1	*Dry Stack Tailings Storage Facility*																		$57,291		
A6.1.1.1		Waste Rock Shell	Regrade WR shell slopes to 4H:1V	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A6.1.1.2			Cover material: Load, haul, dump spread (0.5m)	0	m3	R.048	0.0	$0.86	$0.00	$2.78	$0	$0.91	0	$0.00	$0.00	$0.00	$4.55	$0			
A6.1.1.3			Scarify surface	0	m2	C.2.16	0.0	$0.03	$0.00	$0.07	$0	$0.02	0	$0.00	$0.00	$0.00	$0.12	$0			
A6.1.1.4			Seed/Fertilize: broadcast seeding	0.00	ha	C.6.01	0.0	$1,029.50	$0.00	$401.72	$0	$232.94	0	$0.00	$1,005.72	$0.00	$2,669.89	$0			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Peak Liability Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A6.1.1.5			Tree seedling application (1,000 stems/ha)	5.30	ha	C.6.06	271.4	$1,832.33	$9,703.52	$25.98	$138	$15.00	83	$79.44	$469.71	$2,487.45	$2,343.03	$12,408			
A6.1.2.1		Fill Area at south end of DS	Load, haul, dump, spread	0	Cm3	R.001	0.0	$1.39	$0.00	$3.03	$0	$0.90	0	$0.00	$0.00	$0.00	$5.31	$0			Fill from TDD road and/or unsuitable DSTSF cover material
A6.1.3.1		Tailings surface	Regrade tailings area south of the tailings	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A6.1.3.2			Regrade surface to form tertiary drainage catchments	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A6.1.3.3			Cover: Excavate unsuitable temporary cover material	0	m3	R.003	0.0	$0.51	$0.00	$1.70	$0	$0.57	0	$0.00	$0.00	$0.00	$2.78	$0			
A6.1.3.4			Cover material: Load, haul, dump spread (0.5m)	0	m3	R.048	0.0	$0.86	$0.00	$2.78	$0	$0.91	0	$0.00	$0.00	$0.00	$4.55	$0			
A6.1.3.5			Seed/Fertilize: broadcast seeding	0.00	ha	C.6.01	0.0	$1,029.50	$0.00	$401.72	$0	$232.94	0	$0.00	$1,005.72	$0.00	$2,669.89	$0			
A6.1.3.6			Tree seedling application (1,000 stems/ha)	19.16	ha	C.6.06	981.8	$1,832.33	$35,100.35	$25.98	$498	$15.00	299	$287.34	$469.71	$8,997.81	$2,343.03	$44,883			
A7	**Roads**																			**$136,050**	
A7.1	*Exploration Roads*																		$33,589		
A7.1.1.1		Regrade/Scarify	Scarify road surface	54,285	m2	C.2.16	0.0	$0.03	$1,886.30	$0.07	$3,564	$0.02	965	$926.07	$0.00	$0.00	$0.12	$6,377			
A7.1.2.1		Revegetate	Seed/Fertilize: broadcast seeding	5.43	ha	C.6.01	179.8	$1,029.50	$5,588.65	$401.72	$2,181	$232.94	1,317	$1,264.54	$1,005.72	$5,459.55	$2,669.89	$14,494			
A7.1.2.2			Tree seedling application (1,000 stems/ha)	5.43	ha	C.6.06	278.2	$1,832.33	$9,946.82	$25.98	$141	$15.00	85	$81.43	$469.71	$2,549.82	$2,343.03	$12,719			
A7.2	*Access Roads (excl. Main Site Access Road)*																		$9,082		
A7.2.1.1		Regrade/Scarify	Regrade side slopes	9	hrs	C.2.13	9.5	$44.39	$419.85	$194.68	$1,841	$67.01	660	$633.81	$0.00	$0.00	$306.07	$2,895			
A7.2.1.2			Scarify road surface	52,636	m2	C.2.16	0.0	$0.03	$1,829.00	$0.07	$3,456	$0.02	935	$897.94	$0.00	$0.00	$0.12	$6,183			
A7.2.2.1		Revegetate	Seed/Fertilize: broadcast seeding	0.00	ha	C.6.01	0.0	$1,029.50	$0.77	$401.72	$0	$232.94	0	$0.17	$1,005.72	$0.75	$2,669.89	$2			
A7.2.2.2			Tree seedling application (1,000 stems/ha)	0.00	ha	C.6.06	0.0	$1,832.33	$1.36	$25.98	$0	$15.00	0	$0.01	$469.71	$0.35	$2,343.03	$2			
A7.3	*Main Site Access Road*																		$1,343		
A7.3.1.1		New Access over MVFE	Regrade new access through MVFE	0	m3																
A7.3.2.1		Signage	Install large sign on east and west side of barge landing on Yukon River	2	ea.	C.5.14	1.9	$40.27	$80.54	$15.90	$32	$5.57	12	$11.13	$609.74	$1,219.48	$671.48	$1,343			
A7.4	*Haul Roads*																		$72,436		
A7.4.1.1		Regrade/Scarify	Regrade side slopes	21	hrs	C.2.13	20.5	$44.39	$910.71	$194.68	$3,994	$67.01	1,432	$1,374.82	$0.00	$0.00	$306.07	$6,280			
A7.4.1.2			Scarify road surface	91,084	m2	C.2.16	0.0	$0.03	$3,164.99	$0.07	$5,981	$0.02	1,619	$1,553.84	$0.00	$0.00	$0.12	$10,700			
A7.4.2.1		Revegetate	Seed/Fertilize: broadcast seeding	11.06	ha	C.6.01	366.5	$1,029.50	$11,389.06	$401.72	$4,444	$232.94	2,684	$2,576.99	$1,005.72	$11,125.97	$2,669.89	$29,536			
A7.4.2.2			Tree seedling application (1,000 stems/ha)	11.06	ha	C.6.06	567.0	$1,832.33	$20,270.53	$25.98	$287	$15.00	173	$165.94	$469.71	$5,196.25	$2,343.03	$25,920			
A7.5	*Culvert Removal*																		$19,600		
A7.5.1.1		Excavate culverts	Mobilization allowance for excavation fleet to various sites	6	hrs	C.7.01	12.0	$88.22	$529.34	$118.29	$710	$44.16	276	$264.96	$0.00	$0.00	$250.67	$1,504			
A7.5.1.2			Load, haul, dump, spread (spoil locally)	18,020	m3	R.901	59.1	$0.15	$2,622.79	$0.58	$10,499	$0.20	3,841	$3,687.10	$0.00	$0.00	$0.93	$16,809			
A7.5.1.2		Waste disposal	On-site disposal (demolition debris, etc.)	190	Lm3	R.037	8.1	$1.87	$356.36	$3.75	$713	$1.14	227	$217.54	$0.00	$0.00	$6.77	$1,287			
A8	**Demolition**																			**$1,976,253**	
A8.1	*Airstrip Area*																		$21,059		
A8.1.1.1		Remove tanks/equipment	Dismantle and prep for transport	7	hrs	C.1.13	28.0	$168.00	$1,175.97	$86.19	$603	$14.53	106	$101.70	$0.00	$0.00	$268.71	$1,881			
A8.1.2.1		Waste Oil Tanker Secondar	Bedding Material: Load, haul dump bedding material to landfarm	47	Cm3	R.001	1.5	$1.39	$65.73	$3.03	$143	$0.90	44	$42.57	$0.00	$0.00	$5.31	$252			
A8.1.2.2			Cut and fold liner	219	m2	C.1.15	2.5	$0.43	$94.83	$0.19	$43	$0.07	16	$15.13	$0.00	$0.00	$0.70	$152			
A8.1.2.3			Regrade area to promote positive drainage	192	m2	C.2.11	1.9	$0.44	$85.22	$1.28	$246	$0.35	71	$67.74	$0.00	$0.00	$2.08	$399			
A8.1.3.1		Prepare for demolition	Remove hazardous materials/prep for transport offsite	10	hrs	C.1.22	40.0	$159.30	$1,592.98	$95.83	$958	$26.93	281	$269.28	$0.00	$0.00	$282.05	$2,821			
A8.1.3.2			Disconnect services	4	hrs	C.1.09	12.0	$88.07	$352.28	$7.55	$30	$4.80	20	$19.20	$0.00	$0.00	$100.42	$402			
A8.1.4.1		Demolition	Structural building demolition: wooden buildings/tents	1,149	m3	C.1.08	87.7	$2.88	$3,310.74	$1.60	$1,834	$0.57	680	$652.51	$0.00	$0.00	$5.05	$5,797			
A8.1.4.2			Other demolition: covered storage, debris, etc.	530	m3	C.1.07	52.5	$3.74	$1,980.53	$2.07	$1,097	$0.74	407	$390.34	$0.00	$0.00	$6.55	$3,468			
A8.1.5.1		Waste disposal	On-site disposal (demolition debris, etc.)	379	m3	R.006	12.8	$1.48	$562.06	$2.95	$1,117	$0.92	361	$346.89	$0.00	$0.00	$5.35	$2,026			
A8.1.5.2			Off-site disposal (re-usable equipment, etc.)	4	m3	C.7.07	34.6	$355.10	$1,536.93	$423.68	$1,834	$113.33	511	$490.50	$0.00	$0.00	$892.11	$3,861			
A8.2	*Airport Laydown Area*																		$33,408		
A8.2.1.1		Remove tanks/equipment	Dismantle and prep for transport	2	hrs	C.1.13	8.0	$168.00	$335.99	$86.19	$172	$14.53	30	$29.06	$0.00	$0.00	$268.71	$537			
A8.2.2.1		Prepare for demolition	Remove hazardous materials/prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.2.2.2			Disconnect services	4	hrs	C.1.09	12.0	$88.07	$352.28	$7.55	$30	$4.80	20	$19.20	$0.00	$0.00	$100.42	$402			
A8.2.3.1		Demolition	Structural building demolition: steel structures	96	tonnes	C.1.05	191.3	$69.36	$6,633.00	$139.08	$13,301	$33.84	3,371	$3,236.25	$0.00	$0.00	$242.28	$23,170			
A8.2.3.2			Other demolition: covered storage etc.	560	m3	C.1.07	55.5	$3.74	$2,093.92	$2.07	$1,160	$0.74	430	$412.69	$0.00	$0.00	$6.55	$3,667			
A8.2.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	393	Lm3	R.005	15.8	$1.76	$693.24	$3.53	$1,387	$1.08	441	$423.19	$0.00	$0.00	$6.37	$2,504			
A8.2.4.2			Off-site disposal (re-usable equipment, etc.)	3	trips	C.7.07	23.0	$355.10	$1,020.92	$423.68	$1,218	$113.33	339	$325.82	$0.00	$0.00	$892.11	$2,565			
A8.3	*Camp Area*																		$375,774		
A8.3.1.1		Equipment Removal	Remove salvageable equipment	48	hrs	C.1.13	192.0	$168.00	$8,063.76	$86.19	$4,137	$14.53	726	$697.34	$0.00	$0.00	$268.71	$12,898			
A8.3.2.1		Remove modular buildings	Dismantle and prep for transport	92	ea.	C.1.20	1,794.0	$850.98	$78,290.06	$730.60	$67,215	$166.46	15,953	$15,314.69	$0.00	$0.00	$1,748.04	$160,820			
A8.3.2.2			Transport structures off-site (Whitehorse)	92	ea.	C.7.04	736.0	$351.24	$32,314.23	$303.64	$27,935	$113.33	10,861	$10,426.18	$0.00	$0.00	$768.21	$70,675			
A8.3.3.1		Prepare for demolition	Remove hazardous materials/prep for transport offsite	1	hrs	C.1.22	4.0	$159.30	$159.30	$95.83	$96	$26.93	28	$26.93	$0.00	$0.00	$282.05	$282			
A8.3.3.2			Disconnect services	8	hrs	C.1.09	24.0	$88.07	$704.56	$7.55	$60	$4.80	40	$38.40	$0.00	$0.00	$100.42	$803			
A8.3.4.1		Demolition	Structural building demolition: Steel	175	tonnes	C.1.05	350.7	$69.36	$12,160.50	$139.08	$24,385	$33.84	6,180	$5,933.13	$0.00	$0.00	$242.28	$42,479			
A8.3.4.2			Structural building demolition: Wood/misc. structures	10,959	m3	C.1.08	836.7	$2.88	$31,584.20	$1.60	$17,497	$0.57	6,484	$6,224.86	$0.00	$0.00	$5.05	$55,306			
A8.3.4.4			Other demolition: Utilidors, etc.	466	m3	C.1.07	46.1	$3.74	$1,742.09	$2.07	$965	$0.74	358	$343.34	$0.00	$0.00	$6.55	$3,051			
A8.3.5.1		Waste disposal	On-site disposal (demolition debris, etc.)	3,867	m3	R.007	175.3	$1.99	$7,708.84	$3.99	$15,425	$1.22	4,902	$4,705.86	$0.00	$0.00	$7.20	$27,839			
A8.3.5.2			Off-site disposal (re-usable equipment, etc.)	1	trips	C.7.07	8.0	$355.10	$355.10	$423.68	$424	$113.33	118	$113.33	$0.00	$0.00	$892.11	$892			
A8.3.6.1		Demolish foundations	Break in place concrete foundations	51	m3	C.1.01	4.1	$2.39	$122.15	$9.37	$479	$2.48	132	$126.81	$0.00	$0.00	$14.24	$728			
A8.4	*Explosives Plant and Storage Areas*																		$30,611		
A8.4.1.1		Remove equipment	Small equipment: dismantle and prep for transport	37	hrs	C.1.13	149.3	$168.00	$6,271.81	$86.19	$3,218	$14.53	565	$542.38	$0.00	$0.00	$268.71	$10,032			
A8.4.2.1		Remove modular buildings	Dismantle and prep for transport	1	ea.	C.1.20	19.5	$850.98	$850.98	$730.60	$731	$166.46	173	$166.46	$0.00	$0.00	$1,748.04	$1,748			
A8.4.2.2			Transport structures off-site (Whitehorse)	1	ea.	C.7.04	8.0	$351.24	$351.24	$303.64	$304	$113.33	118	$113.33	$0.00	$0.00	$768.21	$768			
A8.4.3.1		Prepare for demolition	Remove hazardous materials/prep for transport offsite	4	hrs	C.1.22	16.0	$159.30	$637.19	$95.83	$383	$26.93	112	$107.71	$0.00	$0.00	$282.05	$1,128			
A8.4.3.2			Disconnect services	2	hrs	C.1.09	6.0	$88.07	$176.14	$7.55	$15	$4.80	10	$9.60	$0.00	$0.00	$100.42	$201			
A8.4.4.1		Demolition	Structural building demolition: Steel	47	tonnes	C.1.05	94.2	$69.36	$3,266.25	$139.08	$6,550	$33.84	1,660	$1,593.61	$0.00	$0.00	$242.28	$11,410			
A8.4.5.1		Waste disposal	On-site disposal (demolition debris, etc.)	103	m3	R.008	5.4	$2.32	$239.23	$4.65	$479	$1.42	152	$146.04	$0.00	$0.00	$8.39	$864			
A8.4.5.2			Off-site disposal (re-usable equipment, etc.)	5	trips	C.7.07	40.0	$355.10	$1,775.51	$423.68	$2,118	$113.33	590	$566.64	$0.00	$0.00	$892.11	$4,461			
A8.5	*Fuel Storage Area*																		$114,842		
A8.5.1.1		Remove equipment/dismant	Small equipment: dismantle and prep for transport	24	hrs	C.1.13	96.0	$168.00	$4,031.88	$86.19	$2,069	$14.53	363	$348.67	$0.00	$0.00	$268.71	$6,449			
A8.5.1.2			Large equipment (crane req'd): dismantle and prep for transport	192	hrs	C.1.14	1,153.4	$269.89	$51,883.26	$167.56	$32,213	$20.53	4,111	$3,946.31	$0.00	$0.00	$457.98	$88,042			
A8.5.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.5.2.2			Disconnect services	8	hrs	C.1.09	24.0	$88.07	$704.56	$7.55	$60	$4.80	40	$38.40	$0.00	$0.00	$100.42	$803			
A8.5.2.3			Clean out tanks, pressure wash, remove sludge	10	hrs	C.1.04	10.3	$35.54	$365.13	$15.75	$162	$2.00	21	$20.55	$0.00	$0.00	$53.29	$548			
A8.5.3.1		Demolish structures	Misc. debris/scraps (tanks included in dismantling above)	75	m3	C.1.08	5.7	$2.88	$216.15	$1.60	$120	$0.57	44	$42.60	$0.00	$0.00	$5.05	$378			
A8.5.4.1		Remove secondary containr	Bedding Material: Load, haul dump bedding material to landfarm	623	Cm3	R.009	32.8	$2.32	$1,442.73	$4.64	$2,887	$1.41	917	$880.71	$0.00	$0.00	$8.37	$5,210			
A8.5.4.2			Cut and fold liner	1,747	m2	C.1.15	20.3	$0.43	$758.23	$0.19	$340	$0.07	126	$120.97	$0.00	$0.00	$0.70	$1,219			
A8.5.4.3			Regrade area to promote positive drainage	1,666	m2	C.2.11	16.7	$0.44	$739.50	$1.28	$2,138	$0.35	612	$587.76	$0.00	$0.00	$2.08	$3,465			
A8.5.5.1		Waste disposal	On-site disposal (demolition debris, etc.)	229	m3	R.009	12.1	$2.32	$531.18	$4.64	$1,063	$1.41	338	$324.26	$0.00	$0.00	$8.37	$1,918			
A8.5.5.2			Off-site disposal (re-usable equipment, etc.)	7	trips	C.7.07	56.0	$355.10	$2,485.71	$423.68	$2,966	$113.33	826	$793.30	$0.00	$0.00	$892.11	$6,245			
A8.6	*Mill Area*																		$1,265,173		Excludes WTP
A8.6.1.1		Remove equipment	Small equipment: dismantle and prep for transport	379	hrs	C.1.13	1,515.0	$168.00	$63,626.87	$86.19	$32,644	$14.53	5,732	$5,502.37	$0.00	$0.00	$268.71	$101,773			
A8.6.1.2			Large equipment (crane req'd): dismantle and prep for transport	466	hrs	C.1.14	2,796.5	$269.89	$125,789.65	$167.56	$78,099	$20.53	9,966	$9,567.74	$0.00	$0.00	$457.98	$213,456			
A8.6.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	65	hrs	C.1.22	260.0	$159.30	$10,354.37	$95.83	$6,229	$26.93	1,823	$1,750.32	$0.00	$0.00	$282.05	$18,333			Off-site disposal included in 'Waste Disposal'
A8.6.2.6			Reagents: Disposal and tipping fees	1.0	ls	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A8.6.2.7			Decontaminate buildings: wash equipment/structures, etc.	38	hrs	C.1.04	37.8	$35.54	$1,342.93	$15.75	$595	$2.00	79	$75.58	$0.00	$0.00	$53.29	$2,014			
A8.6.2.8			Decontaminate buildings: Wash floors etc.	4,732	m2	C.1.03	40.8	$0.31	$1,448.16	$0.14	$642	$0.02	85	$81.50	$0.00	$0.00	$0.46	$2,171			
A8.6.3.1		Demolition	Structural building demolition: Steel	2,660	tonnes	C.1.05	5,319.1	$69.36	$184,463.18	$139.08	$369,899	$33.84	93,750	$89,999.96	$0.00	$0.00	$242.28	$644,363			
A8.6.3.2			Structural building demolition: Wood/misc. structures	960	m3	C.1.08	73.3	$2.88	$2,766.75	$1.60	$1,533	$0.57	568	$545.29	$0.00	$0.00	$5.05	$4,845			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Peak Liability Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A8.6.3.3			Other demolition: Steel	526	tonnes	C.1.05	1,052.7	$69.36	$36,505.48	$139.08	$73,204	$33.84	18,553	$17,811.10	$0.00	$0.00	$242.28	$127,520			
A8.6.3.4			Other demolition: concrete	602	m3	C.1.06	323.7	$15.92	$9,579.30	$62.48	$37,592	$16.53	10,359	$9,944.51	$0.00	$0.00	$94.92	$57,116			
A8.6.3.5			Other demolition: miscellaneous	244	m3	C.1.07	24.1	$3.74	$910.78	$2.07	$505	$0.74	187	$179.50	$0.00	$0.00	$6.55	$1,595			
A8.6.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	5,985	m3	R.037	255.1	$1.87	$11,217.23	$3.75	$22,445	$1.14	7,133	$6,847.56	$0.00	$0.00	$6.77	$40,510			
A8.6.4.2			Off-site disposal (re-usable equipment, etc.)	33	trips	C.7.07	266.0	$355.10	$11,806.23	$423.68	$14,086	$113.33	3,925	$3,767.87	$0.00	$0.00	$892	$29,660			
A8.6.5.1		Demolish foundations	Break in place concrete foundations	1,532	m3	C.1.01	123.7	$2.39	$3,659.05	$9.37	$14,359	$2.48	3,957	$3,798.55	$0.00	$0.00	$14.24	$21,817			
A8.7	*Mill Valley Fill Extension (Stage 1 and 2)*																		$28,525		
A8.7.1.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	8	hrs	C.1.22	30.0	$159.30	$1,194.73	$95.83	$719	$26.93	210	$201.96	$0.00	$0.00	$282.05	$2,115			
A8.7.1.2			Reagents: Disposal and tipping fees	1	ls	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A8.7.2.1		Demolish buildings	Structural building demolition: Wood/misc. structures (Reagent tent)	1,800	m3	C.1.08	137.4	$2.88	$5,187.66	$1.60	$2,874	$0.57	1,065	$1,022.42	$0.00	$0.00	$5.05	$9,084			
A8.7.3.1		Haul waste to landfill	Demolition debris	594	m3	R.037	25.3	$1.87	$1,113.25	$3.75	$2,228	$1.14	708	$679.58	$0.00	$0.00	$6.77	$4,020			
A8.7.3.2			Remove tires off-site for disposal	10	trips	C.7.07	78.8	$355.10	$3,495.53	$423.68	$4,171	$113.33	1,162	$1,115.57	$0.00	$0.00	$892.11	$8,782			
A8.7.3.3			Misc. debris/scraps	652	m3	R.037	27.8	$1.87	$1,221.95	$3.75	$2,445	$1.14	777	$745.94	$0.00	$0.00	$6.77	$4,413			
A8.7.4.1		Demolish foundations	Break and bury concrete foundations	8	m3	C.1.01	0.6	$2.39	$18.48	$9.37	$73	$2.48	20	$19.19	$0.00	$0.00	$14.24	$110			
A8.8	*Minto South Portal*																		$8,516		
A8.8.1.1		Remove equipment	Small equipment: dismantle and prep for transport	6	hrs	C.1.13	24.0	$168.00	$1,007.97	$86.19	$517	$14.53	91	$87.17	$0.00	$0.00	$268.71	$1,612			
A8.8.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.8.3.1		Demolition	Other demolition: miscellaneous	462	m3	C.1.07	45.7	$3.74	$1,727.03	$2.07	$957	$0.74	355	$340.38	$0.00	$0.00	$6.55	$3,024			
A8.8.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	462	m3	R.043	20.9	$1.99	$918.18	$3.98	$1,837	$1.21	584	$560.51	$0.00	$0.00	$7.18	$3,316			
A8.9	*Minto North Portal*																		$8,981		Qntys assumed to be same as Minto South Portal
A8.9.1.1		Remove equipment	Small equipment: dismantle and prep for transport	6	hrs	C.1.13	24.0	$168.00	$1,007.97	$86.19	$517	$14.53	91	$87.17	$0.00	$0.00	$268.71	$1,612			
A8.9.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.9.3.1		Demolition	Other demolition: miscellaneous	462	m3	C.1.07	45.7	$3.74	$1,727.03	$2.07	$957	$0.74	355	$340.38	$0.00	$0.00	$6.55	$3,024			
A8.9.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	462	m3	R.042	23.8	$2.27	$1,046.71	$4.53	$2,094	$1.38	666	$638.96	$0.00	$0.00	$8.18	$3,780			
A8.10	*Pelly Laydown (Includes propane tanks by W15 Sump)*																		$83,981		
A8.10.1.1		Remove modular buildings/t	Dismantle and prep for transport	7	ea.	C.1.20	136.5	$850.98	$5,956.85	$730.60	$5,114	$166.46	1,214	$1,165.25	$0.00	$0.00	$1,748.04	$12,236			
A8.10.1.2			Transport structures off-site (Whitehorse)	7	ea.	C.7.04	56.0	$351.24	$2,458.69	$303.64	$2,125	$113.33	826	$793.30	$0.00	$0.00	$768.21	$5,377			
A8.10.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.10.3.1		Demolish buildings	Structural building demolition: Steel	80	tonnes	C.1.05	159.4	$69.36	$5,527.50	$139.08	$11,084	$33.84	2,809	$2,696.88	$0.00	$0.00	$242.28	$19,309			
A8.10.3.2			Structural building demolition: Wood/misc. structures	5,672	m3	C.1.08	433.1	$2.88	$16,347.99	$1.60	$9,057	$0.57	3,356	$3,221.99	$0.00	$0.00	$5.05	$28,627			
A8.10.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	2,165	m3	R.045	100.0	$2.03	$4,395.81	$4.06	$8,796	$1.24	2,795	$2,683.42	$0.00	$0.00	$7.33	$15,875			
A8.10.4.2			Off-site disposal (re-usable equipment, etc.)	2	trips	C.7.07	15.0	$355.10	$665.82	$423.68	$794	$113.33	221	$212.49	$0.00	$0.00	$892.11	$1,673			
A8.10.5.1		Demolish foundations	Break and bury concrete foundations	23	m3	C.1.01	1.8	$2.39	$53.73	$9.37	$211	$2.48	58	$55.78	$0.00	$0.00	$14.24	$320			
A8.11	*Vent Raise - Area 118*																		$5,381		
A8.11.1.1		Remove equipment	Small equipment: dismantle and prep for transport	20	hrs	C.1.13	80.0	$168.00	$3,359.90	$86.19	$1,724	$14.53	303	$290.56	$0.00	$0.00	$268.71	$5,374			
A8.11.2.1		Waste disposal	Off-site disposal (re-usable equipment, etc.)	1	trips	R.043	0.0	$1.99	$1.99	$3.98	$4	$1.21	1	$1.21	$0.00	$0.00	$7.18	$7			
A9	**Surface Infrastructure**																			**$169,527**	
A9.1	*Tailings and Water Conveyance Pipelines*																		$91,886		
A9.1.1.1		Dismantle piping systems	Flush and clean tailings pipeline systems	8	ls	C.1.21	32.0	$203.78	$1,630.27	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$203.78	$1,630			
A9.1.1.2			Cut pipelines and prep for transport (6 to 8" HDPE pipes)	6,800	m	C.1.17	1,070.9	$7.21	$49,056.06	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$7.21	$49,056			
A9.1.1.3			Cut pipelines and prep for transport (10 to 18" HDPE pipes)	1,740	m	C.1.18	456.7	$12.02	$20,920.96	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$12.02	$20,921			
A9.1.1.4			Cut pipelines and prep for transport (20 to 36" HDPE pipes)	885	m	C.1.19	348.4	$18.04	$15,961.25	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$18.04	$15,961			
A9.2.2.1		Waste disposal	Haul pipes lines and debris to landfill	638	m3	R.037	27.2	$1.87	$1,195.62	$3.75	$2,392	$1.14	760	$729.86	$0.00	$0.00	$6.77	$4,318			
A9.2	*Conveyance Pipeline Equipment*																		$12,462		
A9.2.1.1		Dismantle piping systems	Remove pumps and prep for transport	15	ea.	C.1.24	198.0	$602.98	$9,044.72	$24.94	$374	$14.40	225	$216.00	$0.00	$0.00	$642.33	$9,635			
A9.2.1.2			Remove barges/Support Equipment prep for transport	2	ea.	C.1.23	52.8	$1,205.96	$2,411.93	$49.89	$100	$28.80	60	$57.60	$0.00	$0.00	$1,284.65	$2,569			
A9.2.2.1		Waste disposal	Transport pumps, barge and support equip. offsite (Whitehorse)	6.7	tonnes	C.7.02	2.7	$17.56	$117.67	$15.18	$102	$5.67	40	$37.96	$0.00	$0.00	$38.41	$257			
A9.3	*Powerlines*																		$65,179		
A9.3.1.1		Dismantle	Dismantle and collect powerlines	3.2	km	C.1.10	217.9	$2,639.96	$8,342.27	$940.09	$2,971	$251.90	829	$795.99	$0.00	$0.00	$3,831.94	$12,109			
A9.3.1.2			Remove power poles and load for transport off-site	60	ea.	C.1.11	548.6	$423.67	$25,420.14	$88.51	$5,311	$18.29	1,143	$1,097.14	$0.00	$0.00	$530.46	$31,828			
A9.3.1.3			Disconnect transformers and load for transport	4	ea.	C.1.12	93.3	$892.89	$3,571.55	$317.96	$1,272	$85.20	355	$340.78	$0.00	$0.00	$1,296.04	$5,184			
A9.3.2.1		Waste disposal	Transport powerlines, poles and transformers off-site	18	trips	C.7.07	144.0	$355.10	$6,391.83	$423.68	$7,626	$113.33	2,125	$2,039.90	$0.00	$0.00	$892.11	$16,058			
A10	**Water Detention Structures**																			**$786,952**	
A10.1	*W15 Sump*																		$1,049		
A10.1.1.1		Remove secondary containr	Cut and fold liner	1,310	m2	C.1.15	15.2	$0.43	$568.50	$0.19	$255	$0.07	94	$90.70	$0.00	$0.00	$0.70	$914			
A10.1.1.2			Haul liner to landfill	20	Lm3	R.070	0.8	$1.90	$37.35	$3.80	$75	$1.16	24	$22.80	$0.00	$0.00	$6.86	$135			
A10.2	*Mill Water Pond*																		$0		
A10.3	*Sewage Lagoon (near IROD)*																		$23,852		
A10.3.1		Remove pond	Backfill pond	4,488	m3	R.001	141.4	$1.39	$6,231.19	$3.03	$13,585	$0.90	4,203	$4,035.32	$0.00	$0.00	$5.31	$23,852			
A10.4	*Water Storage Pond Dam*																		$762,051		
A10.4.1.1		Water Management	Install pump and pump around system	4	hrs	C.4.14	16.0	$120.43	$481.73	$20.53	$82	$6.80	28	$27.20	$0.00	$0.00	$147.76	$591			
A10.4.1.2			Pump pond water to discharge	400,000	m3	C.4.12	146.8	$0.01	$5,216.05	$0.05	$19,052	$0.01	3,058	$2,935.56	$0.00	$0.00	$0.07	$27,203			
A10.4.1.3			Maintain pump around system during WSP closure activities	103	days	C.4.13	10.3	$3.55	$364.31	$12.98	$1,331	$2.00	214	$205.03	$0.00	$0.00	$18.53	$1,900			Duration calculated as sum of associated tasks plus 20% contingency
A10.4.2.1		Breach dam	Granular zones/gen. fill: Load, haul, dump, spread in upstream North ab	62,301	m3	R.001	1,963.4	$1.39	$86,500.02	$3.03	$188,587	$0.90	58,351	$56,017.38	$0.00	$0.00	$5.31	$331,104			
A10.4.2.2			Dam Core: Load, haul, dump in temporary stockpile	23,617	m3	R.001	744.3	$1.39	$32,789.73	$3.03	$71,488	$0.90	22,119	$21,234.62	$0.00	$0.00	$5.31	$125,512			
A10.4.2.3			Rip-rap: Sort and stockpile durable rip-rap for reuse	19,285	m3	R.001	607.8	$1.39	$26,776.14	$3.03	$58,377	$0.90	18,063	$17,340.22	$0.00	$0.00	$5.31	$102,494			
A10.4.2.4			Rip-rap: Load haul, dump unsuitable rip-rap in upstream North abutmen	19,285	m3	R.001	607.8	$1.39	$26,776.14	$3.03	$58,377	$0.90	18,063	$17,340.22	$0.00	$0.00	$5.31	$102,494			
A10.4.3.1		Channel restoration	Excavate stream channel	5,584	m3	R.902	97.1	$0.77	$4,308.58	$1.64	$9,173	$0.46	2,681	$2,574.22	$0.00	$0.00	$2.88	$16,056			
A10.4.3.2			Bedding layer: Load haul, dump, place granular bedding layer along cha	726	m3	R.001	22.9	$1.39	$1,008.60	$3.03	$2,199	$0.90	680	$653.17	$0.00	$0.00	$5.31	$3,861			
A10.4.3.3			Rip-rap: Load, haul dump from temporary stockpile	964	m3	R.001	30.4	$1.39	$1,337.77	$3.03	$2,917	$0.90	902	$866.34	$0.00	$0.00	$5.31	$5,121			
A10.4.3.4			Rip-rap: place and secure	964	m3	R.904	10.5	$0.48	$465.10	$0.91	$877	$0.32	325	$311.83	$0.00	$0.00	$1.72	$1,653			
A10.4.4.1		Cover	Cover material: Load, haul, dump spread (1m)	6,975	m3	R.001	219.8	$1.39	$9,684.17	$3.03	$21,113	$0.90	6,533	$6,271.47	$0.00	$0.00	$5.31	$37,069			Cover sourced from dam core material
A10.4.5.1		Revegetate	Seed/Fertilize: broadcast seeding	1.4	ha	C.6.01	46.2	$1,029.50	$1,436.16	$401.72	$560	$232.94	338	$324.96	$1,005.72	$1,402.98	$2,669.89	$3,724			
A10.4.5.2			Tree seedling application (1,000 stems/ha)	1.4	ha	C.6.06	71.5	$1,832.33	$2,556.10	$25.98	$36	$15.00	22	$20.93	$469.71	$655.25	$2,343.03	$3,269			
A11	**Yards/Laydown Areas**																			**$758,907**	
A11.1	*Airstrip Area*																		$39,834		
A11.1.1.1		Re-grade	Re-grade slopes to be 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A11.1.1.2			Scarify surface	64,378	m2	C.2.16	0.0	$0.03	$2,236.99	$0.07	$4,227	$0.02	1,144	$1,098.24	$0.00	$0.00	$0.12	$7,562			
A11.1.2.1		Revegetate	Seed/Fertilize: broadcast seeding	6.4	ha	C.6.01	213.3	$1,029.50	$6,627.68	$401.72	$2,586	$232.94	1,562	$1,499.64	$1,005.72	$6,474.57	$2,669.89	$17,188			
A11.1.2.2			Tree seedling application (1,000 stems/ha)	6.4	ha	C.6.06	329.9	$1,832.33	$11,796.10	$25.98	$167	$15.00	101	$96.57	$469.71	$3,023.88	$2,343.03	$15,084			
A11.2	*Airport Laydown Area*																		$13,165		
A11.2.1.1		Re-grade	Scarify surface	21,277	m2	C.2.16	0.0	$0.03	$739.33	$0.07	$1,397	$0.02	378	$362.97	$0.00	$0.00	$0.12	$2,499			
A11.2.2.1		Revegetate	Seed/Fertilize: broadcast seeding	2.1	ha	C.6.01	70.5	$1,029.50	$2,190.47	$401.72	$855	$232.94	516	$495.64	$1,005.72	$2,139.87	$2,669.89	$5,681			
A11.2.2.2			Tree seedling application (1,000 stems/ha)	2.1	ha	C.6.06	109.0	$1,832.33	$3,898.65	$25.98	$55	$15.00	33	$31.92	$469.71	$999.40	$2,343.03	$4,985			
A11.3	*Camp Area*																		$118,708		
A11.3.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	7	hrs	C.2.13	6.9	$44.39	$306.20	$194.68	$1,343	$67.01	481	$462.24	$0.00	$0.00	$306.07	$2,111			
A11.3.1.2			Re-grade slopes to be 3H:1V or flatter	24	hrs	C.2.13	24.5	$44.39	$1,086.58	$194.68	$4,766	$67.01	1,709	$1,640.31	$0.00	$0.00	$306.07	$7,493			Slope between camp and main access road
A11.3.1.3			Scarify surfaces	20,626	m2	C.2.16	0.0	$0.03	$716.72	$0.07	$1,354	$0.02	367	$351.87	$0.00	$0.00	$0.12	$2,423			
A11.3.2.1		Cover	Flat area cover material: Load, haul, dump spread	10,313	Cm3	R.046	198.2	$0.85	$8,729.48	$2.75	$28,368	$0.91	9,724	$9,335.37	$0.00	$0.00	$4.50	$46,433			
A11.3.2.2			Slope area cover material: Load, haul, dump along crest	9,349	Cm3	R.047	157.2	$0.74	$6,916.70	$2.44	$22,835	$0.82	7,990	$7,670.53	$0.00	$0.00	$4.00	$37,422			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Peak Liability Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A11.3.2.3		Revegetate	Slope area cover material: Spread down slope	10	hrs	C.2.13	10.2	$44.39	$451.46	$194.68	$1,980	$67.01	710	$681.52	$0.00	$0.00	$306.07	$3,113			
A11.3.3.1			Seed/Fertilize: broadcast seeding	3.93	ha	C.6.01	130.3	$1,029.50	$4,048.49	$401.72	$1,580	$232.94	954	$916.05	$1,005.72	$3,954.97	$2,669.89	$10,499			
A11.3.3.2			Tree seedling application (1,000 stems/ha)	3.93	ha	C.6.06	201.5	$1,832.33	$7,205.60	$25.98	$102	$15.00	61	$58.99	$469.71	$1,847.12	$2,343.03	$9,214			
A11.4	*Crusher Area*																		$73,635		
A11.2.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	9	hrs	C.2.13	9.0	$44.39	$400.28	$194.68	$1,756	$67.01	629	$604.27	$0.00	$0.00	$306.07	$2,760			
A11.2.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	13,482	m3	R.055	244.8	$0.80	$10,783.37	$2.60	$35,042	$0.86	12,012	$11,531.82	$0.00	$0.00	$4.25	$57,358			
A11.2.3.1		Revegetate	Seed/Fertilize: broadcast seeding	2.70	ha	C.6.01	89.3	$1,029.50	$2,775.95	$401.72	$1,083	$232.94	654	$628.11	$1,005.72	$2,711.82	$2,669.89	$7,199			
A11.2.3.2			Tree seedling application (1,000 stems/ha)	3	ha	C.6.06	138.2	$1,832.33	$4,940.70	$25.98	$70	$15.00	42	$40.45	$469.71	$1,266.53	$2,343.03	$6,318			
A11.5	*Exploration Disturbances*																		$21,519		
A11.5.1.1		Ridgetop Area	Scarify surfaces	34,778	m2	C.2.16	0.0	$0.03	$1,208.47	$0.07	$2,284	$0.02	618	$593.29	$0.00	$0.00	$0.12	$4,085			
A11.5.1.2			Seed/Fertilize: broadcast seeding	3	ha	C.6.01	115.2	$1,029.50	$3,580.40	$401.72	$1,397	$232.94	844	$810.13	$1,005.72	$3,497.69	$2,669.89	$9,285			
A11.5.1.3			Tree seedling application (1,000 stems/ha)	3	ha	C.6.06	178.2	$1,832.33	$6,372.49	$25.98	$90	$15.00	54	$52.17	$469.71	$1,633.56	$2,343.03	$8,149			
A11.6	*Explosives Plant and Storage Areas*																		$43,165		
A11.6.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	30	hrs	C.2.13	29.6	$44.39	$1,314.39	$194.68	$5,765	$67.01	2,067	$1,984.22	$0.00	$0.00	$306.07	$9,063			
A11.6.1.2			Re-grade slopes to be 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A11.6.1.3			Scarify surfaces	55,113	m2	C.2.16	0.0	$0.03	$1,915.08	$0.07	$3,619	$0.02	979	$940.20	$0.00	$0.00	$0.12	$6,474			
A11.6.2.1		Revegetate	Seed/Fertilize: broadcast seeding	5.5	ha	C.6.01	182.6	$1,029.50	$5,673.94	$401.72	$2,214	$232.94	1,337	$1,283.83	$1,005.72	$5,542.86	$2,669.89	$14,715			
A11.6.2.2			Tree seedling application (1,000 stems/ha)	5.5	ha	C.6.06	282.5	$1,832.33	$10,098.61	$25.98	$143	$15.00	86	$82.67	$469.71	$2,588.73	$2,343.03	$12,913			
A11.7	*Fuel Storage Areas*																		$24,758		
A11.7.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	5	hrs	C.2.13	4.5	$44.39	$200.33	$194.68	$879	$67.01	315	$302.42	$0.00	$0.00	$306.07	$1,381			
A11.7.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	4,200	m3	R.049	81.8	$0.86	$3,603.19	$2.79	$11,709	$0.92	4,014	$3,853.28	$0.00	$0.00	$4.56	$19,166			
A11.7.3.1		Revegetate	Seed/Fertilize: broadcast seeding	0.84	ha	C.6.01	27.8	$1,029.50	$864.78	$401.72	$337	$232.94	204	$195.67	$1,005.72	$844.80	$2,669.89	$2,243			
A11.7.3.2			Tree seedling application (1,000 stems/ha)	1	ha	C.6.06	43.1	$1,832.33	$1,539.16	$25.98	$22	$15.00	13	$12.60	$469.71	$394.56	$2,343.03	$1,968			
A11.8	*IROD Laydown Area*																		$18,300		
A11.8.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	15	hrs	C.2.13	14.8	$44.39	$654.96	$194.68	$2,873	$67.01	1,030	$988.72	$0.00	$0.00	$306.07	$4,516			
A11.8.1.2			Re-grade slopes to be 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A11.8.1.3			Scarify surface	18,106	m2	C.2.16	0.0	$0.03	$629.15	$0.07	$1,189	$0.02	322	$308.88	$0.00	$0.00	$0.12	$2,127			
A11.8.2.1		Revegetate	Seed/Fertilize: broadcast seeding	2.33	ha	C.6.01	77.0	$1,029.50	$2,393.91	$401.72	$934	$232.94	564	$541.67	$1,005.72	$2,338.60	$2,669.89	$6,208			
A11.8.2.2			Tree seedling application (1,000 stems/ha)	2.33	ha	C.6.06	119.2	$1,832.33	$4,260.73	$25.98	$60	$15.00	36	$34.88	$469.71	$1,092.22	$2,343.03	$5,448			
A11.9	*Mill Area*																		$134,673		
A11.9.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	60	hrs	C.2.13	59.9	$44.39	$2,659.96	$194.68	$11,666	$67.01	4,183	$4,015.50	$0.00	$0.00	$306.07	$18,342			
A11.9.1.2			Scarify surface	42,365	m2	C.2.16	0.0	$0.03	$1,472.09	$0.07	$2,782	$0.02	753	$722.72	$0.00	$0.00	$0.12	$4,977			
A11.9.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	21,182	m3	R.055	384.7	$0.80	$16,942.37	$2.60	$55,057	$0.86	18,873	$18,118.30	$0.00	$0.00	$4.25	$90,118			
A11.9.3.1		Revegetate	Seed/Fertilize: broadcast seeding	4.24	ha	C.6.01	140.3	$1,029.50	$4,361.46	$401.72	$1,702	$232.94	1,028	$986.86	$1,005.72	$4,260.70	$2,669.89	$11,311			
A11.9.3.2			Tree seedling application (1,000 stems/ha)	4.24	ha	C.6.06	217.1	$1,832.33	$7,762.62	$25.98	$110	$15.00	66	$63.55	$469.71	$1,989.91	$2,343.03	$9,926			
A11.10	*Mill Valley Fill Extension (Stage 1 and 2)*																		$0		
A11.10.1		Costs included in Waste Rock Dump Section (WBS#: A1.5)		0	hrs																
A11.11	*Minto South Portal*																		$87,165		
A11.11.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	51	hrs	C.2.13	51.5	$44.39	$2,284.70	$194.68	$10,020	$67.01	3,593	$3,449.00	$0.00	$0.00	$306.07	$15,754			
A11.11.1.2			Re-grade slopes to be 3H:1V or flatter	28	hrs	C.2.13	28.0	$44.39	$1,244.58	$194.68	$5,459	$67.01	1,957	$1,878.82	$0.00	$0.00	$306.07	$8,582			
A11.11.1.3			Scarify surface	62,765	m2	C.2.16	0.0	$0.03	$2,180.95	$0.07	$4,121	$0.02	1,115	$1,070.73	$0.00	$0.00	$0.12	$7,373			
A11.11.2.1		Revegetate	Seed/Fertilize: broadcast seeding	11.1	ha	C.6.01	366.5	$1,029.50	$11,389.08	$401.72	$4,444	$232.94	2,684	$2,576.99	$1,005.72	$11,125.98	$2,669.89	$29,536			
A11.11.2.2			Tree seedling application (1,000 stems/ha)	11.1	ha	C.6.06	567.0	$1,832.33	$20,270.56	$25.98	$287	$15.00	173	$165.94	$469.71	$5,196.26	$2,343.03	$25,920			
A11.12	*Pelly Laydown Area*																		$147,962		
A11.12.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	107	hrs	C.2.13	107.4	$44.39	$4,767.52	$194.68	$20,910	$67.01	7,497	$7,197.08	$0.00	$0.00	$306.07	$32,874			
A11.12.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	25,855	m3	R.058	380.7	$0.65	$16,779.44	$2.11	$54,545	$0.69	18,585	$17,841.31	$0.00	$0.00	$3.45	$89,165			
A11.12.3.1		Revegetate	Seed/Fertilize: broadcast seeding	5.17	ha	C.6.01	171.3	$1,029.50	$5,323.56	$401.72	$2,077	$232.94	1,255	$1,204.55	$1,005.72	$5,200.58	$2,669.89	$13,806			
A11.12.3.2			Tree seedling application (1,000 stems/ha)	5.17	ha	C.6.06	265.0	$1,832.33	$9,474.99	$25.98	$134	$15.00	81	$77.57	$469.71	$2,428.87	$2,343.03	$12,116			
A11.13	*W15 Sump Area Laydown*																		$36,022		
A11.13.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	9	hrs	C.2.13	8.8	$44.39	$390.13	$194.68	$1,711	$67.01	613	$588.94	$0.00	$0.00	$306.07	$2,690			
A11.13.1.2			Slopes: Re-grade to 3H:1V	1	hrs	C.2.13	1.2	$44.39	$55.15	$194.68	$242	$67.01	87	$83.26	$0.00	$0.00	$306.07	$380			
A11.13.1.3			Scarify surface	9,484	m2	C.2.16	0.0	$0.03	$329.54	$0.07	$623	$0.02	169	$161.79	$0.00	$0.00	$0.12	$1,114			
A11.13.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	6,974	m3	R.065	106.1	$0.67	$4,675.54	$2.18	$15,199	$0.71	5,179	$4,971.43	$0.00	$0.00	$3.56	$24,846			
A11.13.3.1		Revegetate	Seed/Fertilize: broadcast seeding	1.39	ha	C.6.01	46.2	$1,029.50	$1,435.97	$401.72	$560	$232.94	338	$324.92	$1,005.72	$1,402.80	$2,669.89	$3,724			
A11.13.3.2			Tree seedling application (1,000 stems/ha)	1.39	ha	C.6.06	71.5	$1,832.33	$2,555.78	$25.98	$36	$15.00	22	$20.92	$469.71	$655.16	$2,343.03	$3,268			
A12	**Waste Disposal**																			**$452,923**	
A12.1	*Hydrocarbon contaminated soils*																		$66,551		
A12.1.1.1		Excavate HC contaminated	Complete testing for contaminated soils	48	test pits	C.3.03	24.0	$22.24	$1,067.53	$2.89	$139	$2.92	146	$139.97	$204.00	$9,792.00	$232.05	$11,138			
A12.1.1.2			Excavate and haul contaminated soils to on-site landfarm facility	726	m3	R.038	26.0	$1.58	$1,144.07	$3.17	$2,298	$0.96	723	$693.78	$0.00	$0.00	$5.70	$4,136			
A12.1.2.1		Construct landfarm	Existing Facility assumed to have sufficient capacity	0	m3																
A12.1.3.1		Operate landfarm	Aerate contaminated soils (mix)	12	event	C.3.04	270.0	$999.20	$11,990.37	$923.10	$11,077	$333.60	4,170	$4,003.20	$0.00	$0.00	$2,255.90	$27,071			
A12.1.3.2			Annual confirmation sampling	3	years	C.3.05	48.0	$711.69	$2,135.06	$92.63	$278	$93.31	292	$279.94	$1,224.00	$3,672.00	$2,121.63	$6,365			
A12.1.4.1		Close landfarm	Remove soils and collect liner	726	m3	R.001	22.9	$1.39	$1,007.57	$3.03	$2,197	$0.90	680	$652.50	$0.00	$0.00	$5.31	$3,857			
A12.1.4.2			Cut/fold liner and place into waste disposal container	4,265	m2	C.1.15	49.6	$0.43	$1,850.89	$0.19	$830	$0.07	308	$295.29	$0.00	$0.00	$0.70	$2,976			
A12.1.4.3			Regrade area	4,265	m2	C.2.11	42.7	$0.44	$1,893.14	$1.28	$5,472	$0.35	1,567	$1,504.69	$0.00	$0.00	$2.08	$8,870			
A12.1.4.4			Revegetation: Seed/Fertilizer: broadcast seeding	0.43	ha	C.6.01	14.1	$1,029.50	$439.08	$401.72	$171	$232.94	103	$99.35	$1,005.72	$428.94	$2,669.89	$1,139			
A12.1.4.5			Tree seedling application (1,000 stems/ha)	0.43	ha	C.6.06	21.9	$1,832.33	$781.49	$25.98	$11	$15.00	7	$6.40	$469.71	$200.33	$2,343.03	$999			
	Metal Contaminated soils																		$159,118		
A12.2.1.1		Crusher area	Delineate contaminated soil areas	197	test pits	C.3.01	98.5	$22.24	$4,381.32	$2.89	$570	$2.92	598	$574.45	$265.20	$52,244.40	$293.25	$57,770			
A12.2.1.2			Load, haul, dump contaminated soils to underground	2,949	m3	R.039	102.1	$1.53	$4,499.64	$3.33	$9,810	$0.99	3,035	$2,913.97	$0.00	$0.00	$5.84	$17,224			
A12.2.1.3			Complete confirmation testing	197	ea	C.3.02	32.8	$7.43	$1,463.48	$1.33	$261	$0.00	0	$0.00	$13.26	$2,612.22	$22.01	$4,337			
A12.2.2.1		Mill Area	Delineate contaminated soil areas	198	test pits	C.3.01	99.0	$22.24	$4,403.56	$2.89	$573	$2.92	601	$577.37	$265.20	$52,509.60	$293.25	$58,064			
A12.2.2.2			Load, haul, dump contaminated soils to underground	2,973	m3	R.039	103.0	$1.53	$4,536.49	$3.33	$9,890	$0.99	3,060	$2,937.83	$0.00	$0.00	$5.84	$17,365			
A12.2.2.3			Complete confirmation testing	198	ea	C.3.02	33.0	$7.43	$1,470.91	$1.33	$262	$0.00	0	$0.00	$13.26	$2,625.48	$22.01	$4,359			
A12.3	*Solid Waste Landfill*																		$217,134		
A12.3.1.1		Construction landfill	Place waste from site facilities	12,920	LCM	C.2.07	550.6	$1.89	$24,440.80	$8.30	$107,194	$2.86	38,433	$36,895.99	$0.00	$0.00	$13.04	$168,531			
A12.3.1.2			Place fill to minimize voids in the debris	3,371	m3	R.001	106.2	$1.39	$4,679.67	$3.03	$10,203	$0.90	3,157	$3,030.55	$0.00	$0.00	$5.31	$17,913			
A12.3.2.1		Close landfill	Soil Cover: Load haul, dump spread, compact (0.6m)	5,024	m3	R.002	190.0	$1.61	$8,100.50	$3.21	$16,144	$0.99	5,196	$4,988.33	$0.00	$0.00	$5.82	$29,233			
A12.3.2.2			Revegetation: Seed/Fertilizer: broadcast seeding	0.55	ha	C.6.01	18.1	$1,029.50	$562.06	$401.72	$219	$232.94	132	$127.18	$1,005.72	$549.07	$2,669.89	$1,458			
A12.4	*Hazardous Material Off-Site Disposal*																		$10,120		
A12.4.1.1		Hazardous Waste	From Building Demo: Transport off-site for disposal	2	trip	C.7.04	14.5	$351.24	$634.94	$303.64	$549	$113.33	213	$204.86	$0.00	$0.00	$768.21	$1,389			
A12.4.3.3			Hazardous materials: Disposal and Tipping fees.	1	ls	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0	$0			
A12.4.2.1		Reagents	Reagents: Transport to Whitehorse	25	hrs	C.1.22	99.3	$159.30	$3,955.90	$95.83	$2,380	$26.93	697	$668.71	$0.00	$0.00	$282.05	$7,004			
A12.4.2.2			Reagents: Disposal and tipping fees	2.2	trips	C.7.04	18.0	$351.24	$789.50	$303.64	$682	$113.33	265	$254.73	$0.00	$0.00	$768.21	$1,727			
A13	**Surface Water Conveyance**																			**$3,287,496**	
A13.1	*Channel A - W-15 to Main Pit*																		$207,036		
A13.1.1.1		Excavate channel	Load, haul, dump locally	40,300	m3	R.902	700.5	$0.77	$31,093.04	$1.64	$66,199	$0.46	19,351	$18,576.93	$0.00	$0.00	$2.88	$115,869			
A13.1.2.1		Place channel materials	Geotextile: Supply and place	7,785	m3	C.5.08	221.4	$1.09	$8,520.21	$0.11	$854	$0.11	897	$860.72	$2.16	$16,802.77	$3.47	$27,038			
A13.1.2.2			Rip-rap (angular, high quality): Screen and stockpile	3,855	m3	C.2.15	0.0	$2.54	$9,777.29	$6.06	$23,344	$1.86	7,462	$7,163.19	$0.00	$0.00	$10.45	$40,285			
A13.1.2.3			Rip-rap: Load, haul, dump	3,855	m3	R.018	102.2	$1.17	$4,501.31	$2.55	$9,814	$0.76	3,037	$2,915.04	$0.00	$0.00	$4.47	$17,230			
A13.1.2.4			Rip-rap: Place and secure	3,855	m3	R.904	41.9	$0.48	$1,860.68	$0.91	$3,507	$0.32	1,299	$1,247.51	$0.00	$0.00	$1.72	$6,615			
A13.2	*Channel B - W-35 to Area 2 Pit*																		$198,304		

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Peak Liability Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A13.2.1.1		Excavate channel	Load, haul, dump locally	48,038	m3	R.902	835.0	$0.77	$37,063.29	$1.64	$78,910	$0.46	23,067	$22,143.92	$0.00	$0.00	$2.88	$138,117			
A13.2.2.1		Place channel materials	Geotextile: Supply and place	4,682	m2	C.5.08	133.1	$1.09	$5,124.06	$0.11	$514	$0.11	539	$517.64	$2.16	$10,105.20	$3.47	$16,261			
A13.2.2.2			Rip-rap (angular, high quality): Screen and stockpile	2,554	m3	C.2.15	0.0	$2.54	$6,477.46	$6.06	$15,465	$1.86	4,943	$4,745.61	$0.00	$0.00	$10.45	$26,689			
A13.2.2.3			Rip-rap: Load, haul, dump	2,554	m3	R.020	76.2	$1.32	$3,358.44	$2.87	$7,322	$0.85	2,266	$2,174.93	$0.00	$0.00	$5.03	$12,855			
A13.2.2.4			Rip-rap: Place and secure	2,554	m3	R.904	27.8	$0.48	$1,232.70	$0.91	$2,323	$0.32	861	$826.47	$0.00	$0.00	$1.72	$4,382			
A13.3	*Channel C - Main Pit to Main Access Road*																		$125,034		
A13.3.2.1		Excavate channel	Load, haul, dump locally	8,656	m3	R.902	150.5	$0.77	$6,678.68	$1.64	$14,219	$0.46	4,157	$3,990.26	$0.00	$0.00	$2.88	$24,888			
A13.3.3.1		Intake structure	Excavate soils to competent foundation materials	2,450	Bm3	R.902	42.6	$0.77	$1,890.07	$1.64	$4,024	$0.46	1,176	$1,129.25	$0.00	$0.00	$2.88	$7,043			
A13.3.3.2			Supply an install precast-concrete intake wingwall intake structure	1	ls	C.5.12	140.0	$5,946	$5,945.57	$3,583	$3,583	$643.84	671	$643.84	$37,000	$37,000.00	$47,172	$47,172			
A13.3.3.6			Backfill and compact around structure	2,450	Cm3	C.2.04	237.3	$2.58	$6,323.25	$3.01	$7,367	$1.03	2,638	$2,532.57	$0.00	$0.00	$6.62	$16,223			
A13.3.4.1		Place channel materials	Geotextile: Supply and place	3,171	m2	C.5.08	90.2	$1.09	$3,470.31	$0.11	$348	$0.11	365	$350.57	$2.16	$6,843.83	$3.47	$11,013			
A13.3.4.2			Rip-rap (angular, high quality): Screen and stockpile	1,037	m3	C.2.15	0.0	$2.54	$2,631.28	$6.06	$6,282	$1.86	2,008	$1,927.76	$0.00	$0.00	$10.45	$10,841			
A13.3.4.3			Rip-rap: Load, haul, dump	1,037	m3	R.022	36.4	$1.54	$1,600.27	$3.33	$3,452	$0.98	1,064	$1,021.09	$0.00	$0.00	$5.85	$6,073			
A13.3.4.4			Rip-rap: Place and secure	1,037	m3	R.904	11.3	$0.48	$500.75	$0.91	$944	$0.32	350	$335.73	$0.00	$0.00	$1.72	$1,780			
A13.4	*Channel D - Area 2 Pit to Channel E*																		$195,636		
A13.4.1.1		Excavate channel	Load, haul, dump locally	17,840	m3	R.902	310.1	$0.77	$13,763.89	$1.64	$29,304	$0.46	8,566	$8,223.41	$0.00	$0.00	$2.88	$51,291			
A13.4.2.1		Intake structure	Excavate soils to competent foundation materials	4,453	Bm3	R.902	77.4	$0.77	$3,435.48	$1.64	$7,314	$0.46	2,138	$2,052.57	$0.00	$0.00	$2.88	$12,802			
A13.4.2.2			Supply an install precast-concrete intake wingwall intake structure	1	ls	C.5.12	140.0	$5,946	$5,945.57	$3,583	$3,583	$643.84	671	$643.84	$37,000	$37,000.00	$47,172	$47,172			
A13.4.2.6			Backfill and compact around structure	4,453	Cm3	C.2.04	431.4	$2.58	$11,493.42	$3.01	$13,390	$1.03	4,795	$4,603.31	$0.00	$0.00	$6.62	$29,487			
A13.4.3.1		Place channel materials	Geotextile: Supply and place	4,910	m2	C.5.08	139.6	$1.09	$5,373.23	$0.11	$539	$0.11	565	$542.81	$2.16	$10,596.59	$3.47	$17,051			
A13.4.3.2			Rip-rap (angular, high quality): Screen and stockpile	2,199	m3	C.2.15	0.0	$2.54	$5,578.78	$6.06	$13,320	$1.86	4,258	$4,087.21	$0.00	$0.00	$10.45	$22,986			
A13.4.3.3			Rip-rap: Load, haul, dump	2,199	m3	R.020	65.7	$1.32	$2,892.50	$2.87	$6,306	$0.85	1,951	$1,873.18	$0.00	$0.00	$5.03	$11,072			
A13.4.3.4			Rip-rap: Place and secure	2,199	m3	R.904	23.9	$0.48	$1,061.68	$0.91	$2,001	$0.32	741	$711.81	$0.00	$0.00	$1.72	$3,774			
A13.5	*Channel E - Main Access Road to toe of MVFE*																		$739,690		
A13.5.1.1		Construct channel	Bulk fill to grade: Load, haul, dump, spread	29,750	m3																
A13.5.1.2			Excavate/form Channel: Load, haul, dump locally	46,386	m3	R.001	1,461.8	$1.39	$64,402.28	$3.03	$140,409	$0.90	43,445	$41,706.89	$0.00	$0.00	$5.31	$246,519			
A13.5.2.1		Place channel materials	Geotextile: Supply and place	27,307	m2	C.5.08	776.5	$1.09	$29,885.47	$0.11	$2,997	$0.11	3,145	$3,019.05	$2.16	$58,937.36	$3.47	$94,839			
A13.5.2.2			Rip-rap (angular, high quality): Screen and stockpile	20,558	m3	C.2.15	0.0	$2.54	$52,143.07	$6.06	$124,496	$1.86	39,794	$38,201.83	$0.00	$0.00	$10.45	$214,841			
A13.5.2.3			Rip-rap: Load, haul, dump	20,558	m3	R.025	845.7	$1.81	$37,226.38	$3.91	$80,296	$1.16	24,743	$23,753.12	$0.00	$0.00	$6.87	$141,275			
A13.5.2.4			Rip-rap: Place and secure	20,558	m3	R.904	223.6	$0.48	$9,923.17	$0.91	$18,701	$0.32	6,930	$6,653.05	$0.00	$0.00	$1.72	$35,277			
A13.5.3.1		Stilling basin/Energy Dissipa	Excavate basin (spoil locally)	645	m3	R.902	11.2	$0.77	$497.87	$1.64	$1,060	$0.46	310	$297.46	$0.00	$0.00	$2.88	$1,855			
A13.5.3.2			Geotextile: Supply and place	540	m2	C.5.08	15.4	$1.09	$591.07	$0.11	$59	$0.11	62	$59.71	$2.16	$1,165.65	$3.47	$1,876			
A13.5.3.3			Bedding layer: Screen and stockpile	131	m3	C.2.01	2.7	$0.96	$125.01	$2.11	$276	$0.44	60	$57.55	$0.00	$0.00	$3.51	$459			
A13.5.3.4			Bedding layer: Load, haul, dump and place	131	m3	R.028	7.7	$2.59	$337.69	$5.78	$755	$1.69	229	$220.12	$0.00	$0.00	$10.05	$1,312			
A13.5.3.5			Rip-rap (at inlet and outlet): Screen and stockpile	75	m3	C.2.15	0.0	$2.54	$190.23	$6.06	$454	$1.86	145	$139.37	$0.00	$0.00	$10.45	$784			
A13.5.3.6			Rip-rap (at inlet and outlet): Load, haul, dump	75	m3	R.027	3.2	$1.86	$139.15	$3.97	$298	$1.17	91	$87.84	$0.00	$0.00	$7.00	$525			
A13.5.3.7			Rip-rap (at inlet and outlet): Place and secure	75	m3	R.904	0.8	$0.48	$36.20	$0.91	$68	$0.32	25	$24.27	$0.00	$0.00	$1.72	$129			
A13.6	*Minto Creek Wetland By-Pass Channel*																		$231,721		
A13.6.1.1		Excavate channel	Load, haul, dump locally	29,668	m3	R.902	515.7	$0.77	$22,889.69	$1.64	$48,733	$0.46	14,246	$13,675.73	$0.00	$0.00	$2.88	$85,299			
A13.6.2.1		Place channel materials	Geotextile: Supply and place	12,212	m2	C.5.08	347.3	$1.09	$13,364.91	$0.11	$1,340	$0.11	1,406	$1,350.13	$2.16	$26,357.04	$3.47	$42,412			
A13.6.2.2			Rip-rap (angular, high quality): Screen and stockpile	5,950	m3	C.2.15	0.0	$2.54	$15,091.24	$6.06	$36,032	$1.86	11,517	$11,056.37	$0.00	$0.00	$10.45	$62,179			
A13.6.2.3			Rip-rap: Load, haul, dump	5,950	m3	R.001	187.5	$1.39	$8,260.73	$3.03	$18,010	$0.90	5,573	$5,349.64	$0.00	$0.00	$5.31	$31,620			Assumes WSP rip-rap can be re-used
A13.6.2.4			Rip-rap: Place and secure	5,950	m3	R.904	64.7	$0.48	$2,871.96	$0.91	$5,412	$0.32	2,006	$1,925.52	$0.00	$0.00	$1.72	$10,210			
A13.7	*Allowance: Minto North Spillway*																		$83,528		
A13.7.1.1		Excavate channel	Load, haul, dump locally	13,035	m3	R.902	226.6	$0.77	$10,056.98	$1.64	$21,412	$0.46	6,259	$6,008.67	$0.00	$0.00	$2.88	$37,477			
A13.7.2.1		Place channel materials	Geotextile: Supply and place	4,136	m2	C.5.08	117.6	$1.09	$4,526.14	$0.11	$454	$0.11	476	$457.23	$2.16	$8,926.04	$3.47	$14,363			
A13.7.2.2			Rip-rap (angular, high quality): Screen and stockpile	1,943	m3	C.2.15	0.0	$2.54	$4,929.10	$6.06	$11,769	$1.86	3,762	$3,611.23	$0.00	$0.00	$10.45	$20,309			
A13.7.2.3			Rip-rap: Load, haul, dump	1,943	m3	R.024	47.7	$1.08	$2,101.36	$2.36	$4,581	$0.70	1,418	$1,360.84	$0.00	$0.00	$4.14	$8,044			Assumes WSP rip-rap can be re-used
A13.7.2.4			Rip-rap: Place and secure	1,943	m3	R.904	21.1	$0.48	$938.04	$0.91	$1,768	$0.32	655	$628.91	$0.00	$0.00	$1.72	$3,335			
A13.8	*Ridgetop North Outlet Channel*																		$74,863		
A13.8.1.1		Excavate channel	Load, haul, dump locally	7,758	m3	R.902	134.9	$0.77	$5,985.73	$1.64	$12,744	$0.46	3,725	$3,576.25	$0.00	$0.00	$2.88	$22,306			
A13.8.2.1		Place channel materials	Geotextile: Supply and place	3,995	m2	C.5.08	113.6	$1.09	$4,371.69	$0.11	$438	$0.11	460	$441.63	$2.16	$8,621.43	$3.47	$13,873			
A13.8.2.2			Rip-rap (angular, high quality): Screen and stockpile	2,531	m3	C.2.15	0.0	$2.54	$6,419.56	$6.06	$15,327	$1.86	4,899	$4,703.20	$0.00	$0.00	$10.45	$26,450			
A13.8.2.3			Rip-rap: Load, haul, dump	2,531	m3	R.068	33.1	$0.58	$1,455.99	$1.90	$4,821	$0.64	1,681	$1,613.82	$0.00	$0.00	$3.12	$7,891			Assumes WSP rip-rap can be re-used
A13.8.2.4			Rip-rap: Place and secure	2,531	m3	R.904	27.5	$0.48	$1,221.69	$0.91	$2,302	$0.32	853	$819.09	$0.00	$0.00	$1.72	$4,343			
A13.9	*Secondary Catchment Channels*																		$791,748		
A13.9.1.1		Southwest Dump	Excavate mild-graded channels (good access), spoil locally	10,787	Bm3	R.902	187.5	$0.77	$8,322.78	$1.64	$17,720	$0.46	5,180	$4,972.55	$0.00	$0.00	$2.88	$31,015			
A13.9.1.2			Excavate steep-graded channels (poor access), spoil locally	6,640	Bm3	R.903	131.0	$0.88	$5,815.42	$1.86	$12,381	$0.52	3,619	$3,474.49	$0.00	$0.00	$3.26	$21,671			
A13.9.1.3			Geotextile: Supply and place	25,206	m2	C.5.08	716.7	$1.09	$27,585.63	$0.11	$2,766	$0.11	2,903	$2,786.72	$2.16	$54,401.82	$3.47	$87,541			
A13.9.1.4			Bedding layer: Screen and stockpile	2,054	Cm3	C.2.01	41.8	$0.96	$1,967.27	$2.11	$4,343	$0.44	943	$905.70	$0.00	$0.00	$3.51	$7,216			
A13.9.1.5			Bedding layer: Load, haul, dump and place in steep areas	2,054	Cm3	R.032	72.2	$1.55	$3,176.63	$3.34	$6,852	$0.99	2,111	$2,026.92	$0.00	$0.00	$5.87	$12,055			
A13.9.1.6			Rip-rap (angular, high quality): Screen and stockpile	6,806	Cm3	C.2.15	0.0	$2.54	$17,263.99	$6.06	$41,219	$1.86	13,175	$12,648.20	$0.00	$0.00	$10.45	$71,131			
A13.9.1.7			Rip-rap: Load, haul, dump	6,806	m3	R.031	223.5	$1.45	$9,846.77	$3.15	$21,468	$0.94	6,642	$6,376.76	$0.00	$0.00	$5.54	$37,691			
A13.9.1.8			Rip-rap: Place in mild-graded channels (good access)	3,787	m3	R.904	41.2	$0.48	$1,828.10	$0.91	$3,445	$0.32	1,277	$1,225.66	$0.00	$0.00	$1.72	$6,499			
A13.9.1.9			Rip-rap: Place in steep-graded channels (poor access)	3,019	m3	R.905	74.5	$1.10	$3,308.57	$2.33	$7,044	$0.65	2,059	$1,976.75	$0.00	$0.00	$4.08	$12,329			
A13.9.2.1		Main Waste Dump	Excavate mild-graded channels (good access), spoil locally	6,477	Bm3	R.902	112.6	$0.77	$4,997.00	$1.64	$10,639	$0.46	3,110	$2,985.52	$0.00	$0.00	$2.88	$18,621			
A13.9.2.2			Excavate steep-graded channels (poor access), spoil locally	3,986	Bm3	R.903	78.6	$0.88	$3,490.67	$1.86	$7,432	$0.52	2,172	$2,085.55	$0.00	$0.00	$3.26	$13,008			
A13.9.2.3			Geotextile: Supply and place	15,132	m2	C.5.08	430.3	$1.09	$16,560.93	$0.11	$1,661	$0.11	1,743	$1,672.99	$2.16	$32,659.93	$3.47	$52,555			
A13.9.2.4			Bedding layer: Screen and stockpile	1,233	Cm3	C.2.01	25.1	$0.96	$1,180.84	$2.11	$2,607	$0.44	566	$543.64	$0.00	$0.00	$3.51	$4,331			
A13.9.2.5			Bedding layer: Load, haul, dump and place in steep areas	1,233	Cm3	R.030	34.8	$1.25	$1,535.65	$2.76	$3,401	$0.82	1,059	$1,016.39	$0.00	$0.00	$4.83	$5,953			
A13.9.2.6			Rip-rap (angular, high quality): Screen and stockpile	4,086	Cm3	C.2.15	0.0	$2.54	$10,364.11	$6.06	$24,745	$1.86	7,909	$7,593.11	$0.00	$0.00	$10.45	$42,702			
A13.9.2.7			Rip-rap: Load, haul, dump	4,086	m3	R.029	101.4	$1.09	$4,473.09	$2.42	$9,908	$0.72	3,084	$2,960.56	$0.00	$0.00	$4.24	$17,341			
A13.9.2.8			Rip-rap: Place in mild-graded channels (good access)	2,274	m3	R.904	24.7	$0.48	$1,097.59	$0.91	$2,068	$0.32	767	$735.89	$0.00	$0.00	$1.72	$3,902			
A13.9.2.9			Rip-rap: Place in steep-graded channels (poor access)	1,812	m3	R.905	44.7	$1.10	$1,985.96	$2.33	$4,228	$0.65	1,236	$1,186.53	$0.00	$0.00	$4.08	$7,401			
A13.9.3.1		Reclamation Overburden Du	Excavate mild-graded channels (good access), spoil locally	4,886	Bm3	R.902	84.9	$0.77	$3,769.89	$1.64	$8,026	$0.46	2,346	$2,252.36	$0.00	$0.00	$2.88	$14,049			
A13.9.3.2			Excavate steep-graded channels (poor access), spoil locally	3,007	Bm3	R.903	59.3	$0.88	$2,633.47	$1.86	$5,607	$0.52	1,639	$1,573.40	$0.00	$0.00	$3.26	$9,814			
A13.9.3.3			Geotextile: Supply and place	11,416	m2	C.5.08	324.6	$1.09	$12,494.07	$0.11	$1,253	$0.11	1,315	$1,262.16	$2.16	$24,639.65	$3.47	$39,649			
A13.9.3.4			Bedding layer: Screen and stockpile	930	Cm3	C.2.01	18.9	$0.96	$890.86	$2.11	$1,967	$0.44	427	$410.14	$0.00	$0.00	$3.51	$3,268			
A13.9.3.5			Bedding layer: Load, haul, dump and place in steep areas	930	Cm3	R.034	34.7	$1.64	$1,527.43	$3.58	$3,330	$1.06	1,030	$989.17	$0.00	$0.00	$6.29	$5,847			
A13.9.3.6			Rip-rap (angular, high quality): Screen and stockpile	3,083	Cm3	C.2.15	0.0	$2.54	$7,819.00	$6.06	$18,669	$1.86	5,967	$5,728.47	$0.00	$0.00	$10.45	$32,216			
A13.9.3.7			Rip-rap: Load, haul, dump	3,083	m3	R.033	97.0	$1.39	$4,271.96	$3.02	$9,314	$0.90	2,882	$2,766.52	$0.00	$0.00	$5.30	$16,352			
A13.9.3.8			Rip-rap: Place in mild-graded channels (good access)	1,715	m3	R.904	18.7	$0.48	$828.06	$0.91	$1,561	$0.32	578	$555.18	$0.00	$0.00	$1.72	$2,944			
A13.9.3.9			Rip-rap: Place in steep-graded channels (poor access)	1,367	m3	R.905	33.8	$1.10	$1,498.27	$2.33	$3,190	$0.65	932	$895.16	$0.00	$0.00	$4.08	$5,583			
A13.9.4.1		DSTSF & MVFE	Excavate mild-graded channels (good access), spoil locally	7,378	Bm3	R.902	128.3	$0.77	$5,692.78	$1.64	$12,120	$0.46	3,543	$3,401.22	$0.00	$0.00	$2.88	$21,214			
A13.9.4.2			Excavate steep-graded channels (poor access), spoil locally	4,541	Bm3	R.903	89.6	$0.88	$3,976.72	$1.86	$8,467	$0.52	2,475	$2,375.94	$0.00	$0.00	$3.26	$14,819			
A13.9.4.3			Geotextile: Supply and place	17,239	m2	C.5.08	490.2	$1.09	$18,866.88	$0.11	$1,892	$0.11	1,985	$1,905.94	$2.16	$37,207.52	$3.47	$59,872			
A13.9.4.4			Bedding layer: Screen and stockpile	1,405	Cm3	C.2.01	28.6	$0.96	$1,345.27	$2.11	$2,970	$0.44	645	$619.34	$0.00	$0.00	$3.51	$4,935			
A13.9.4.5			Bedding layer: Load, haul, dump and place in steep areas	1,405	Cm3	R.028	82.5	$2.59	$3,633.88	$5.78	$8,121	$1.69	2,467	$2,368.79	$0.00	$0.00	$10.05	$14,124			
A13.9.4.6			Rip-rap (angular, high quality): Screen and stockpile	4,655	Cm3	C.2.15	0.0	$2.54	$11,807.22	$6.06	$28,191	$1.86	9,011	$8,650.38	$0.00	$0.00	$10.45	$48,648			
A13.9.4.7			Rip-rap: Load, haul, dump	4,655	m3	R.027	196.3	$1.86	$8,636.41	$3.97	$18,485	$1.17	5,679	$5,451.80	$0.00	$0.00	$7.00	$32,573			
A13.9.4.8			Rip-rap: Place in mild-graded channels (good access)	2,590	m3	R.904	28.2	$0.48	$1,250.42	$0.91	$2,356	$0.32	873	$838.35	$0.00	$0.00	$1.72	$4,445			
A13.9.4.9			Rip-rap: Place in steep-graded channels (poor access)	2,065	m3	R.905	51.0	$1.10	$2,262.48	$2.33	$4,817	$0.65	1,408	$1,351.75	$0.00	$0.00	$4.08	$8,431			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Peak Liability Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A13.9.5.1		Ridgetop Waste Dump	Excavate mild-graded channels (good access), spoil locally	574	Bm3	R.902	10.0	$0.77	$442.77	$1.64	$943	$0.46	276	$264.54	$0.00	$0.00	$2.88	$1,650			
A13.9.5.2			Excavate steep-graded channels (poor access), spoil locally	353	Bm3	R.903	7.0	$0.88	$309.30	$1.86	$659	$0.52	192	$184.80	$0.00	$0.00	$3.26	$1,153			
A13.9.5.3			Geotextile: Supply and place	876	m2	C.5.08	24.9	$1.09	$958.89	$0.11	$96	$0.11	101	$96.87	$2.16	$1,891.03	$3.47	$3,043			
A13.9.5.4			Bedding layer: Screen and stockpile	465	Cm3	C.2.01	9.4	$0.96	$444.99	$2.11	$982	$0.44	213	$204.86	$0.00	$0.00	$3.51	$1,632			
A13.9.5.5			Bedding layer: Load, haul, dump and place in steep areas	109	Cm3	R.063	2.9	$1.18	$128.79	$3.83	$419	$1.26	143	$137.73	$0.00	$0.00	$6.27	$685			
A13.9.5.6			Rip-rap (angular, high quality): Screen and stockpile	362	Cm3	C.2.15	0.0	$2.54	$918.34	$6.06	$2,193	$1.86	701	$672.81	$0.00	$0.00	$10.45	$3,784			
A13.9.5.7			Rip-rap: Load, haul, dump	362	m3	R.064	8.5	$1.03	$373.01	$3.40	$1,231	$1.14	431	$413.67	$0.00	$0.00	$5.57	$2,018			
A13.9.5.8			Rip-rap: Place in mild-graded channels (good access)	201	m3	R.904	2.2	$0.48	$97.26	$0.91	$183	$0.32	68	$65.21	$0.00	$0.00	$1.72	$346			
A13.9.5.9			Rip-rap: Place in steep-graded channels (poor access)	161	m3	R.905	4.0	$1.10	$175.97	$2.33	$375	$0.65	110	$105.14	$0.00	$0.00	$4.08	$656			
A13.10	*Tailings Diversion Ditch*																		$624,970		
A13.10.1.2		Regrade	Slopes: Re-grade to 3H:1V	28	hrs	C.2.13	28	$44.39	$1,226	$194.68	$5,376	$67.01	1,928	$1,850	$0.00	$0	$306.07	$8,452			
A13.10.2.1		Cover	Flat area cover material: Load, haul, dump spread	25,182	Cm3	R.066	562.3	$0.98	$24,763	$3.20	$80,473	$1.05	27,585	$26,482	$0.00	$0	$5.23	$131,718			
A13.10.2.2			Slope area cover material: Load, haul, dump along crest	18,049	Cm3	R.067	352.6	$0.86	$15,513	$2.84	$51,216	$0.95	17,921	$17,204	$0.00	$0	$4.65	$83,933			
A13.10.2.3			Slope area cover material: Spread down slope	14	hrs	C.2.13	14.0	$44.39	$621	$194.68	$2,724	$67.01	977	$937	$0.00	$0	$306.07	$4,282			
A13.10.3.1		Revegetate	Seed/Fertilize: tractor application	9	ha	C.6.01	286.4	$1,029.50	$8,901	$401.72	$3,473	$232.94	2,098	$2,014	$1,005.72	$8,696	$2,670	$23,085			
A13.10.3.2			Tree seedling application (1,000 stems/ha)	9	ha	C.6.06	443.1	$1,832.33	$15,843	$25.98	$225	$15.00	135	$130	$469.71	$4,061	$2,343	$20,258			
A13.10.4.1		Extend Channel to WSP	Clear and grub footprint	13,889	m2	C.2.03	52.8	$0.17	$2,344	$0.49	$6,774	$0.13	1,940	$1,863	$0.00	$0	$0.79	$10,981			
A13.10.4.2			Geotextile: Supply and place	13,889	m2	C.5.08	394.9	$1.09	$15,200	$0.11	$1,524	$0.11	1,600	$1,536	$2.16	$29,976	$3.47	$48,236			
A13.10.4.3			Rip-rap (angular, high quality): Screen and stockpile	13,169	m3	C.2.15	0.0	$2.54	$33,403	$6.06	$79,753	$1.86	25,492	$24,472	$0.00	$0	$10.45	$137,629			
A13.10.4.4			Rip-rap: Load, haul, dump	13,169	m3	R.035	618.4	$2.07	$27,207	$4.42	$58,233	$1.30	17,890	$17,175	$0.00	$0	$7.79	$102,615			
A13.10.4.5			Rip-rap: Place in steep-graded channels (poor access)	13,169	m3	R.905	325.1	$1.10	$14,432	$2.33	$30,726	$0.65	8,982	$8,622	$0.00	$0	$4.08	$53,780			
A14	**Water Treatment**																			**$483,085**	
A14.1	*Passive Treatment System - Minto Creek Valley*																		$397,994		
A14.1.1.1		Water Management	Install pump and pump around system	8	hrs	C.4.14	32.0	$120.43	$963.46	$20.53	$164	$6.80	57	$54.40	$0.00	$0.00	$147.76	$1,182			Assumed
A14.1.1.2			Maintain pump around system during wetland construction activities	1,440	hrs	C.4.13	144.0	$3.55	$5,117.32	$12.98	$18,691	$2.00	3,000	$2,880.00	$0.00	$0.00	$18.53	$26,689			Assumes wetlands constructed over 2 months
A14.1.2.1		Valley Fill placement	Load, haul, dump material from the WSP excavation	85,918	m3	n/a - costed	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A14.1.2.1			Load, haul dump valley fill material	0	m3	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			Fill from WSP excavation exceeds the required amount.
A14.1.3.1		Construct wetlands	Excavate cells: Load, haul dump, spoil locally	14,459	m3	R.902	251.3	$0.77	$11,155.61	$1.64	$23,751	$0.46	6,943	$6,665.06	$0.00	$0.00	$2.88	$41,572			
A14.1.3.2			Subgrade preparation: remove sharp objects, place fill as required	9,273	m2	C.2.18	317.0	$1.31	$12,142.93	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$1.31	$12,143			
A14.1.3.3			Geotextile: Supply and install	10,201	m2	C.5.08	290.1	$1.09	$11,163.84	$0.11	$1,120	$0.11	1,175	$1,127.78	$2.16	$22,016.30	$3.47	$35,428			
A14.1.3.4			HDPE liner: Supply and install	10,201	m2	C.5.11	1,187.0	$4.74	$48,319.12	$0.27	$2,749	$0.27	2,884	$2,769.07	$9.38	$95,723.03	$14.66	$149,560			
A14.1.3.5			Organics: Mulch organics, produce and stockpile	5,122	Cm3	C.2.10	577.1	$4.50	$23,023.50	$4.87	$24,922	$1.82	9,720	$9,331.25	$0.00	$0.00	$11.18	$57,277			
A14.1.3.6			Organics: Load, haul, dump, place in cell	5,122	Cm3	R.001	161.4	$1.39	$7,111.43	$3.03	$15,504	$0.90	4,797	$4,605.36	$0.00	$0.00	$5.31	$27,221			
A14.1.3.7			Plant wetland vegetation	0.9	ha	C.6.05	40.6	$1,443.27	$1,338.39	$307.03	$285	$180.00	174	$166.92	$1,159.23	$1,074.99	$3,089.53	$2,865			
A14.1.4.1		Conveyance channels	Excavate channels: Load, haul, dump locally	1,402	m3	R.902	24.4	$0.77	$1,081.90	$1.64	$2,303	$0.46	673	$646.39	$0.00	$0.00	$2.88	$4,032			
A14.1.4.2			Surface preparation: remove sharp objects, place fill as required	1,858	m2	C.2.18	63.5	$1.31	$2,432.63	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$1.31	$2,433			
A14.1.4.3			Geotextile: Supply and place	1,858	m2	C.5.08	52.8	$1.09	$2,033.17	$0.11	$204	$0.11	214	$205.39	$2.16	$4,009.63	$3.47	$6,452			
A14.1.4.4			HDPE Liner: Supply and install	1,858	m2	C.5.11	216.2	$4.74	$8,799.92	$0.27	$501	$0.27	525	$504.31	$9.38	$17,433.16	$14.66	$27,238			
A14.1.4.5			Bedding/Protection layer: Screen and stockpile	442	m3	C.2.01	9.0	$0.96	$423.54	$2.11	$935	$0.44	203	$194.99	$0.00	$0.00	$3.51	$1,554			
A14.1.4.6			Bedding/Protection layer: Load, haul, dump and place	442	m3	R.001	13.9	$1.39	$614.04	$3.03	$1,339	$0.90	414	$397.65	$0.00	$0.00	$5.31	$2,350			Assumes WSP fill can be re-used
A14.1.5.1		Operation and maintenance	System to be started Yr. 1 post-closure	0	yr	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A14.2	*Southwest Wetlands*																		$15,944		
A14.2.1		Berm construction	Construct 3 berms across valley	3,000	m3	R.001	94.5	$1.39	$4,165.24	$3.03	$9,081	$0.90	2,810	$2,697.41	$0.00	$0.00	$5.31	$15,944			Assumed
A14.2.2		Outlet structures	Excavate outlet channels through berms	1	ls																Allowance
A14.3	*Decommission/demolish Water Treatment System*																		$69,147		
A14.3.1.1		Remove equipment	Small equipment: dismantle and prep for transport	17	hrs	C.1.13	67.1	$168.00	$2,819.22	$86.19	$1,446	$14.53	254	$243.80	$0.00	$0.00	$268.71	$4,509			
A14.3.1.2			Large equipment (crane req'd): dismantle and prep for transport	29	hrs	C.1.14	175.7	$269.89	$7,903.93	$167.56	$4,907	$20.53	626	$601.18	$0.00	$0.00	$457.98	$13,412			
A14.3.2.1		Prepare for demolition	Reagents: Load for transport	25	hrs	C.1.22	99.3	$159.30	$3,955.90	$95.83	$2,380	$26.93	697	$668.71	$0.00	$0.00	$282.05	$7,004			
A14.3.2.2			Reagents: Transport to Whitehorse	2.2	trips	C.7.04	18.0	$351.24	$789.50	$303.64	$682	$113.33	265	$254.73	$0.00	$0.00	$768.21	$1,727			
A14.3.2.3			Reagents: Disposal and tipping fees	1.0	ls	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A14.3.3.1		Demolition	Structural building demolition	163	tonnes	C.1.05	325.0	$69.36	$11,272.12	$139.08	$22,604	$33.84	5,729	$5,499.69	$0.00	$0.00	$242.28	$39,376			
A14.3.3.2			Other demolition: miscellaneous	15	m3	C.1.07	1.5	$3.74	$56.09	$2.07	$31	$0.74	12	$11.05	$0.00	$0.00	$6.55	$98			
A14.3.4.1		Waste disposal	Off-site disposal (re-usable equipment, etc.)	3	trips	C.7.07	27.1	$355.10	$1,202.32	$423.68	$1,435	$113.33	400	$383.71	$0.00	$0.00	$892	$3,021			
Subtotal Direct Costs - Active Closure							**125,316**		**$5,342,835**		**$9,935,632**			**$3,059,285**		**$1,093,950**				**$19,431,701**	

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	Peak Liability Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
INDIRECT COSTS																					
B1	**Mobilization-Demobilization**																			**$324,077**	
B1.1		Mobilization	Year 0	1	LS												$162,039	$162,038.53			
B1.2		Demobilization	Year 3 (end of Active Closure)	1	LS												$144,406	$144,405.50			
B1.4		Demobilization	End of Passive closure	1	LS												$17,633	$17,633.03			
B2	**Transportation Costs**																			**$419,142**	
B2.1.1		Barge Operations		15	month												$10,832	$162,486			
		Staffing Bus trips during barge operation period		60	ea (one per week)												$678	$40,656			Turnaround labour costs included in labour build-up
B2.2.2		Air transport and airstrip operations		72	flights												$3,000	$216,000			Three flights per week
B3	**Site/Road Maintenance**																			**$791,662**	
B3.1	*Road Maintenance*																		$727,887		
B3.1.1		Water Truck	Assume needed 150 hrs per month	21	month												$12,153	$255,205			
B3.1.2		Grader	Assume needed 150 hrs per month (16H + Operator)	21	month												$22,509	$472,682			Includes Main Access Road throughout closure period.
B3.2	*Soil Erosion*																		$63,775		
B3.2.1		Erosion control allowance	Silt fencing: supply, install remove at strategic locations	7024	m	C.2.17	283.5	$3.23	$22,661.96	$1.93	$13,533	$0.68	4,941	$4,743.08	$0.92	$6,485.58	$6.75	$47,424			
B3.2.2			Supply and install erosion control matting	10000	m2	C.2.09	175.4	$0.66	$6,622.75	$0.09	$912	$0.05	548	$526.32	$0.83	$8,290.76	$1.64	$16,352			
B4	**Construction Support**																			**$5,344,480**	
B4.1	*Field Support Staff*																		$3,235,218		Turnaround labour costs included in labour build-up
B4.1.1		Mine Manager		21	month			(360) site working hours per month									$23,408	$491,578			
B4.1.2		Office/Camp manager		21	month			(360) site working hours per month									$22,267	$467,614			
B4.1.3		Security/Administrative Assistant		21	month			(360) site working hours per month									$16,600	$348,591			
B4.1.4		Foreman		21	month			(360) site working hours per month									$17,769	$373,152			
B4.1.5		Mechanic		21	month			(360) site working hours per month									$17,769	$373,152			
B4.1.6		Surveyor		21	month			(360) site working hours per month									$17,769	$373,152			
B4.1.7		Engineering technician	Material & QA/QC testing	10.5	month			(360) site working hours per month									$42,120	$442,260			Assumed needed 1/2 of time
B4.1.8		Medic/H&S supervisor		21	month			(360) site working hours per month									$17,415	$365,718			
B4.2	*Field support Vehicles*																		$862,785		
B4.2.1		Pick-up trucks (4 required)		84	month												$4,446	$373,452			
B4.2.2		Fuel truck		21	month												$4,720	$99,123			
B4.2.3		Mechanic service vehicle		21	month												$11,803	$247,873			
B4.2.4		Emergency transport vehicle		21	month												$4,519	$94,901			
B4.2.5		Passenger bus		21	month												$2,259	$47,436			
B4.3	*Field Support Equipment/Supplies*																		$1,246,477		
B4.3.1		Light Towers		21	month			2 # of Units									$3,356	$70,476			
B4.3.2		Material/Laboratory testing allowance		21	month												$1,000	$21,000			
B4.4		Construction, Environmental, and H&S Management Plans		1	LS												$25,000	$25,000			
B4.5		Office supplies		21	month												$1,000	$21,000			
B4.6		Communications		21	month												$1,000	$21,000			
B4.7		Misc. supplies		21	month												$500	$10,500			
B4.8		Camp costs	Includes catering and housekeeping	12,532	man-day												$80	$1,002,531			
B4.9		Power and heat		21	month												$3,570	$74,970			
B5	**QA and Project Management**																			**$1,752,126**	
B5.1		Project Manager		21	month			(160) site working hours per month									$9,897	$207,828			
B5.2		Design Engineer		21	month			(360) site working hours per month									$48,600	$1,020,600			
B5.3		Environmental Monitor		21	month			(360) site working hours per month									$16,046	$336,972			
B5.4		Engineering , Design, and Construction plans (included in Planning and Permitting Costs)															$0.00	$0			
B5.5		Field support vehicles (2 vehicles)		42	month												$4,446	$186,726			
Subtotal Indirect Costs - Active Closure																				**$8,631,488**	
CLOSURE IMPLEMENTATION COSTS - TOTAL																				**$28,063,189**	

Worksheet 10 - Peak Liability Estimate - Schedule Details

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

A: Year 0 Site Access Schedule

Stage	Flight Months	Helicopter Months	Barge Months	Total Months	Month												Comments/Notes
					Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	
Interim Care	2	5	5	12	H	H	H	F	F	B	B	B	B	B	H	H	
SUB-TOTAL INTERIM CARE	**2**	**5**	**5**	**12**													
Active Closure 1	2	0	5	7				F	F	B	B	B	B	B			
Active Closure 2	2	0	5	7				F	F	B	B	B	B	B			
Active Closure 3	2	0	5	7				F	F	B	B	B	B	B			
SUB-TOTAL ACTIVE CLOSURE	**6**	**0**	**15**	**21**													
Post-Closure 1	2	0	3	5				F	F	B	B	B					
SUB-TOTAL POST-CLOSURE 1	**2**	**0**	**3**	**5**													
Post-Closure 2	0	0	1	1						B							
SUB-TOTAL POST-CLOSURE 2	**0**	**0**	**1**	**1**													

B: Annual C&M Staffing Schedule - Interim Operations

#	Role		Months Required	Month												Comments/Notes
				Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	
Months Active at Site			**12**	**1**	**1**	**1**	**1**	**1**	**1**	**1**	**1**	**1**	**1**	**1**	**1**	
Administration/Office																
1	Mine Manager		4				0.5	0.5	0.5	0.5	0.5	0.5	0.5			
2	Office/Camp Mgr		6	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	
3	Payroll/Accounting/HR		4				0.5	0.5	0.5	0.5	0.5	0.5	0.5			
Water Treatment/Environmental Staffing																
1	Warehouse/Water Treatment Operator		5				1.0	1.0	1.0	1.0	1.0					
2	Environmental Manager		4				1.0	0.5	0.5	0.5	0.5	0.5	0.5			
3	Environmental Technician		7	0.5	0.5	0.5	1.0	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	
Operations																
4	Barge Operator		10						2.0	2.0	2.0	2.0	2.0			
5	Equipment operators		13				2.0	3.0	1.5	1.5	1.5	1.5	1.5			
6	HD Mechanic		2				0.5	0.3	0.3	0.3	0.25	0.25	0.25			
7	Tradesmen		2				0.5	0.3	0.3	0.3	0.25	0.25	0.5			
8	Labour/Helpers		10	1	1	1.0	1.0	1.0	0.5	0.5	0.5	0.5	0.5	1.0	1.0	
Camp/Support																
1	Cooks/Housecleaning etc.		0				0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0		Contract (accounted for in camp costs)
Other																
1	Visitors		2				0.25	0.25	0.25	0.25	0.25	0.25	0.25	0.0		
TOTAL			**66.5**	**2.0**	**2.0**	**2.0**	**8.8**	**8.3**	**8.3**	**8.3**	**8.3**	**7.3**	**7.5**	**2.0**	**2.0**	
		Annual man-days:	**1,995**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)
2. Orange highlighted cells are used for camp-man day calculations only.

C: Annual C&M Staffing Schedule - During Active Phase

#	Role		Months Required	Month												Comments/Notes
				Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	
Months Active at Site			**7**				**1**	**1**	**1**	**1**	**1**	**1**	**1**			
Water Treatment/Environmental Staffing																
1	Warehouse/Water Treatment Operator		5				1.0	1.0	1.0	1.0	1.0					
2	Environmental Manager		4				1.0	0.5	0.5	0.5	0.5	0.5	0.5			
3	Environmental Technician		4				1.0	0.5	0.5	0.5	0.5	0.5	0.5			
TOTAL			**13**	**0**	**0**	**0**	**3**	**2**	**2**	**2**	**2**	**1**	**1**	**0**	**0**	
		Annual man-days:	**390**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)

D: Annual C&M Staffing Schedule - Post Closure 1

#	Role		Months Required	Month: Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	Comments/Notes
Months Active at Site			**5**				**1**	**1**	**1**	**1**	**1**					
Administration/Office																
1	Mine Manager		3				0.5	0.5	0.5	0.5	0.5					
2	Office/Camp Mgr		3				1.0	0.5	0.5	0.5	0.5					
3	Payroll/Accounting/HR		1				0.25	0.25	0.25	0.25	0.25					
Water Treatment/Environmental Staffing																
1	Water Treatment Operator		2				0.5	0.5	0.3	0.3	0.3					
2	Environmental Manager		3				0.5	0.5	0.5	0.5	0.5					
3	Environmental Technician		3				0.5	0.5	0.5	0.5	0.5					
Operations																
4	Barge Operator		6						2.0	2.0	2.0					
5	Equipment operators		10				2.0	3.0	1.5	1.5	1.5					
6	HD Mechanic		2				1.0	0.25	0.25	0.25	0.25					
7	Tradesmen		2				1.0	0.25	0.25	0.25	0.25					
8	Labour/Helpers		2				1.0	0.25	0.25	0.25	0.5					
Camp/Support																
1	Cooks/Housecleaning etc.		0				0.0	0.0	0.0	0.0	0.0					Contract (accounted for in camp costs)
Other																
1	Visitors		1				0.25				0.25					
TOTAL			**36**	**0**	**0**	**0**	**9**	**7**	**7**	**7**	**7**	**0**	**0**	**0**	**0**	
	Annual man-days:		**1,073**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)
2. Orange highlighted cells are used for camp-man day calculations only.

E: Annual C&M Staffing Schedule - Post Closure 2

#	Role		Months Required	Month: Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	Comments/Notes
Months Active at Site			**1**						**1**							
Administration/Office																
1	Mine Manager		1						0.5							
2	Office/Camp Mgr		0						0.0							
3	Payroll/Accounting/HR		0						0.00							
Water Treatment/Environmental Staffing																
1	Passive Treatment Scientist		1						0.5							
2	Environmental Manager		0													
3	Environmental Technician		0													
Operations																
4	Barge Operator		1						1.0							
5	Equipment operators		1						1.0							
6	HD Mechanic		1						0.50							
7	Tradesmen		0													
8	Labour/Helpers		0						0.25							
Camp/Support																
1	Cooks/Housecleaning etc.		0						0.0							Contract (accounted for in camp costs)
Other																
1	Visitors		0						0.25							
TOTAL			**4**	**0**	**0**	**0**	**0**	**0**	**4**	**0**	**0**	**0**	**0**	**0**	**0**	
	Annual man-days:		**120**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)
2. Orange highlighted cells are used for camp-man day calculations only.

E: Annual C&M Staffing Schedule - Long-term perpetual maintenance year

#	Role		Months Required	Month: Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	Comments/Notes
Months Active at Site			**1**							**1**						
Administration/Office																
1	Project Manager/Foreman		1							1.0						
Water Treatment/Engineering Staffing																
1	Passive Treatment Scientist		1							1.0						
2	Site Engineer		1							1.0						
3	Environmental Technician		0													
Operations																
4	Barge Operator		1							1.0						Assumed required 1/2 the time (near mob and demob)
5	Equipment operators		4							4.0						
6	HD Mechanic		1							0.50						
7	Tradesmen		0													
8	Labour/Helpers		1							1.00						
Camp/Support																
1	Cooks/Housecleaning etc.		0													Contract (accounted for in camp costs)
Other																
1	Visitors		1							0.50						
TOTAL			**10**	**0**	**0**	**0**	**0**	**0**	**0**	**10**	**0**	**0**	**0**	**0**	**0**	
	Annual man-days:		**300**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)
2. Orange highlighted cells are used for camp-man day calculations only.

Worksheet 11- Peak Liability Estimate - Planning and Permitting Costs

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

Planning And Permitting

WBS	Facility/Area	Task	Activity	Qnty	Unit	Total Unit Rate ($/unit)	Activity Total	Subtotal	Source / Comments
M1	**Planning and Permitting**								
M1.1	*Reclamation Research/Planning*							$400,000	
M1.1.1		Complete reclamation closure and research plan		2	yr	$200,000	$400,000		
M1.2	*Technical studies and investigations*							$58,500	
M1.2.1		Allowance for additional technical studies		1	ls	$58,500	$58,500		
M1.3	*Monitoring and Management Plans*							$209,158	
M1.3.1		Adaptive Mgmt Plans	Physcial, water quality, etc	1	ls	$15,000	$15,000		
M1.3.2		Revegetation plan		1	ls	$30,000	$30,000		
M1.3.3		Waste Management Plan	Water treatment, sludge, landfarm, etc.	1	ls	$50,000	$50,000		
M1.3.4		MMER Final Monitoring Study		1	ls	$114,158	$114,158		
M1.4	*Engineering, Design, and Construction Plans*							$971,585	
M1.4.1		Percentage of direct implementation costs		5%	of	$19,431,701	$971,585		
M1.5	*Permitting*							$0	
M1.5.1.1		Permit Staffing	Permitting Manager	0	ls	$0	$0		Staffing costs included in tasks above
M1.5.1.2			Environmental Manager	0	ls	$0	$0		
M1.5.1.3			Technical Consultants	0	ls	$0	$0		
TOTAL								**$1,639,243**	

Worksheet 12 - End-of-Mine Liability Estimate - Cashflow Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

NPV CALCULATION INPUTS

NOTE: Columns between Years 2049 and 2128 are hidden for printing purposes.

Input Parameters

Scenario:	EOM - 2022	
Current Year:	2018	Year that the estimate is costed
Closure Year 1:	2022	
Closure Period:	3	yrs
Post-Closure Year 1:	2025	

Short-Term Interest Rate (Interim Care/Implementation Costs)

##	Inflation Rate	Years in effect	Start Year	End Year
1	2.0%	100	2018	2118

Long-Term Monitoring/Maintenance NPV Discount Rates

##	Discount Rate	Years in effect	Start Year	End Year
1	2.35%	2	2018	2020
2	2.5%	3	2021	2024
3	2.8%	-	2025	

SCHEDULE

Implementation Schedule

	Area	Closure Year:	OPERATIONS				CLOSURE			POST CLOSURE																											
			Year -4	Year -3	Year -2	Year -1	Year 1	Year 2	Year 3	Year 4	Year 5	Year 6	Year 7	Year 8	Year 9	Year 10	Year 11	Year 12	Year 13	Year 14	Year 15	Year 16	Year 17	Year 18	Year 19	Year 20	Year 21	Year 22	Year 23	Year 24	Year 25	Year 26	Year 27	Year 28	Year 107	Year 108
		Sum	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2128	2129
Direct Costs	Waste Dumps	100%					40%	60%																												
	Overburden Dumps	100%							100%																											
	Ore Stockpiles	100%					100%																													
	Open Pits	100%						50%	50%																											
	Underground Openings	100%					100%																													
	External Tailings Facilities	100%					50%	50%																												
	Roads	100%							100%																											
	Demolition	100%					10%	75%	15%																											
	Surface Infrastructure	100%					25%	50%	25%																											
	Water Detention Structures	100%						50%	50%																											
	Yards/Laydown Areas	100%					25%	25%	50%																											
	Waste Disposal	100%						12%	75%	2%	2%	2%	6%																							
	Surface Water Conveyance	100%					20%	30%	50%																											
	Water Treatment	100%							86%					14%																						
Indirect	Mobilization-Demobilization	100%					50%		45%					5%																						
	Transport Costs	100%					34%	33%	33%																											
	Road Maintenance	100%					34%	33%	33%																											
	Construction Support	100%					34%	33%	33%																											
	QA and Pro. Mgmt.	100%					34%	33%	33%																											

Monitoring Schedule

			Set Monitoring/C&M Stage:				2 - Active	2 - Active	2 - Active	3 - PC 1	3 - PC 1	3 - PC 1	3 - PC 1	3 - PC 1	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2	4 - PC 2									5 - Perpetual		5 - Perpetual
		Post-Closure Year:	Year -7	Year -6	Year -5	Year -4	Year -3	Year -2	Year -1	Year 1	Year 2	Year 3	Year 4	Year 5	Year 6	Year 7	Year 8	Year 9	Year 10	Year 11	Year 12	Year 13	Year 14	Year 15	Year 16	Year 17	Year 18	Year 19	Year 20	Year 21	Year 22	Year 23	Year 24	Year 25	Year 104	Year 105
		Sum																																		
Planning and permitting	Reclamation Research and Planning	100%			50%	50%																													
	Technical Studies and Investigations	100%			50%	50%																													
	Engineering, Design, and Construction Plans	100%			30%	40%	10%	10%	10%																											
	Monitoring and Management Plans	100%			50%	50%																													
	Permitting	100%			50%	50%																													
Monitoring	Water Quality Monitoring (surface and groundwater)						1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1										1		1
	Sediment Monitoring						1	1	1	1		1		1					1					1										1		1
	Biological Monitoring						1	1	1	1		1		1					1					1										1		1
	Geotechnical Monitoring						1	1	1	1	1	1	1	1					1					1										1		1
	Revegetation Monitoring						1	1	1	1	1	1	1	1					1					1										1		1
	Annual Inspection Reporting						1	1	1	1	1	1	1	1					1					1										1		1

Care and Maintenance Schedule

Water Treatment/Management	Active System						x	x	x	x	x	x	x	x	x	x	x	x	x	x	x	x	x	x												
	Passive System									x	x	x	x	x					x					x										x		x
Reclamation Maintenance	Revegetation, cover repair, etc.									x	x	x	x	x					x														x		x
Site Maintenance							x	x	x	x	x	x	x	x					x					x										x		x
On-site Management							x	x	x	x	x	x	x	x					x					x										x		x
Transportation Costs							x	x	x	x	x	x	x	x					x					x										x		x

UNDISCOUNTED CASH FLOW (TO 100 YEARS POST CLOSURE)

Implementation Costs

	Area		Total	% of Total	Check Sum	Year -4	Year -3	Year -2	Year -1	Year 1	Year 2	Year 3	Year 4	Year 5	Year 6	Year 7	Year 8	Year 9	Year 10	Year 11	Year 12	Year 13	Year 14	Year 15	Year 16	Year 17	Year 18	Year 19	Year 20	Year 21	Year 22	Year 23	Year 24	Year 25	Year 26	Year 27	Year 28	Year 107	Year 108
						2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2128	2129
Direct Costs	Waste Dumps		$6,987,586	27%	$6,987,586	$0	$0	$0	$0	$2,795,034	$4,192,552	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Overburden Dumps		$395,276	2%	$395,276	$0	$0	$0	$0	$0	$0	$395,276	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Ore Stockpiles		$794,200	3%	$794,200	$0	$0	$0	$0	$794,200	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Open Pits		$620,582	2%	$620,582	$0	$0	$0	$0	$0	$310,291	$310,291	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Underground Openings		$81,637	0%	$81,637	$0	$0	$0	$0	$81,637	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	External Tailings Facilities		$57,291	0%	$57,291	$0	$0	$0	$0	$28,646	$28,646	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Roads		$136,050	1%	$136,050	$0	$0	$0	$0	$0	$0	$136,050	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Demolition		$1,976,253	8%	$1,976,253	$0	$0	$0	$0	$197,625	$1,482,189	$296,438	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Surface Infrastructure		$169,527	1%	$169,527	$0	$0	$0	$0	$42,382	$84,763	$42,382	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Water Detention Structures		$786,952	3%	$786,952	$0	$0	$0	$0	$0	$393,476	$393,476	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Yards/Laydown Areas		$758,907	3%	$758,907	$0	$0	$0	$0	$189,727	$189,727	$379,454	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Waste Disposal		$452,923	2%	$452,923	$0	$0	$0	$0	$0	$54,828	$339,692	$11,145	$11,145	$11,145	$24,967	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Surface Water Conveyance		$3,287,496	13%	$3,287,496	$0	$0	$0	$0	$657,499	$986,249	$1,643,748	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Water Treatment		$483,085	2%	$483,085	$0	$0	$0	$0	$0	$0	$413,938	$0	$0	$0	$0	$69,147	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Indirect	Mobilization-Demobilization		$324,077	1%	$324,077	$0	$0	$0	$0	$162,039	$0	$144,406	$0	$0	$0	$0	$17,633	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Transport Costs		$419,142	2%	$419,142	$0	$0	$0	$0	$142,508	$138,317	$138,317	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Site/Road Maintenance		$791,662	3%	$791,662	$0	$0	$0	$0	$269,165	$261,249	$261,249	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Construction Support		$5,263,751	21%	$5,263,751	$0	$0	$0	$0	$1,789,675	$1,737,038	$1,737,038	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	QA and Pro. Mgmt.		$1,752,126	7%	$1,752,126	$0	$0	$0	$0	$595,723	$578,202	$578,202	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Subtotal - Closure Implementation Costs			**$25,538,523**		$25,538,523	**$0**	**$0**	**$0**	**$0**	**$7,745,860**	**$10,437,526**	**$7,209,955**	**$11,145**	**$11,145**	**$11,145**	**$24,967**	**$86,780**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

Monitoring Costs

				Total	% of Total	Check Sum	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2128	2129
1	Planning and	Reclamation Research and Planning		$200,000	6%	$200,000	$0	$0	$100,000	$100,000	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2	Permitting	Technical Studies and Investigations		$58,500	2%	$58,500	$0	$0	$29,250	$29,250	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3		Engineering, Design, and Construction Plans		$849,388	24%	$849,388	$0	$0	$254,816	$339,755	$84,939	$84,939	$84,939	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4		Monitoring and Management Plans		$209,158	6%	$209,158	$0	$0	$104,579	$104,579	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5		Permitting		$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
1	Monitoring	Water Quality Monitoring		$1,172,423	33%	$1,172,423	$0	$0	$0	$0	$130,437	$130,437	$130,437	$51,987	$51,987	$51,987	$51,987	$51,987	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$27,220	$0	$0	$0	$0	$0	$0	$0	$0	$27,664	$0	$27,664
2		Sediment Monitoring		$57,596	2%	$57,596	$0	$0	$0	$0	$3,388	$3,388	$3,388	$3,388	$0	$3,388	$0	$3,388	$0	$0	$0	$0	$3,388	$0	$0	$0	$0	$3,388	$0	$0	$0	$0	$0	$0	$0	$0	$3,388	$0	$3,388
3		Biological Monitoring		$208,999	6%	$208,999	$0	$0	$0	$0	$12,294	$12,294	$12,294	$12,294	$0	$12,294	$0	$12,294	$0	$0	$0	$0	$12,294	$0	$0	$0	$0	$12,294	$0	$0	$0	$0	$0	$0	$0	$0	$12,294	$0	$12,294
4		Geotechnical Monitoring		$354,653	10%	$354,653	$0	$0	$0	$0	$12,409	$12,409	$12,409	$12,409	$12,409	$12,409	$12,409	$12,409	$0	$0	$0	$0	$23,216	$0	$0	$0	$0	$23,216	$0	$0	$0	$0	$0	$0	$0	$0	$23,216	$0	$23,216
5		Revegetation Monitoring		$200,071	6%	$200,071	$0	$0	$0	$0	$10,530	$10,530	$10,530	$10,530	$10,530	$10,530	$10,530	$10,530	$0	$0	$0	$0	$10,530	$0	$0	$0	$0	$10,530	$0	$0	$0	$0	$0	$0	$0	$0	$10,530	$0	$10,530
6		Annual Inspection Reporting		$281,876	8%	$281,876	$0	$0	$0	$0	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$2,372	$0	$0	$0	$0	$23,900	$0	$0	$0	$0	$23,900	$0	$0	$0	$0	$0	$0	$0	$0	$23,900	$0	$23,900
Post Closure Costs Subtotal				**$3,592,664**		$3,592,664	**$0**	**$0**	**$488,645**	**$573,584**	**$256,369**	**$256,369**	**$256,369**	**$92,981**	**$77,299**	**$92,981**	**$77,299**	**$92,981**	**$27,220**	**$27,220**	**$27,220**	**$27,220**	**$100,548**	**$27,220**	**$27,220**	**$27,220**	**$27,220**	**$100,548**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$100,993**	**$0**	**$100,993**

Care and Maintenance Costs

				Total	% of Total	Check Sum	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2128	2129
1	Water Treatment	Active System		$6,567,079	39%	$6,567,079	$0	$0	$0	$0	$706,967	$706,967	$706,967	$706,967	$706,967	$706,967	$706,967	$706,967	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$91,135	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Passive System		$560,239	3%	$560,239	$0	$0	$0	$0	$0	$0	$0	$35,015	$35,015	$35,015	$35,015	$35,015	$0	$0	$0	$0	$0	$35,015	$0	$0	$0	$0	$35,015	$0	$0	$0	$0	$0	$0	$0	$0	$35,015	$0	$35,015
3	Reclamation Maintenance	(Cover/Reveg Repairs)		$1,386,205	8%	$1,386,205	$0	$0	$0	$0	$0	$0	$0	$277,241	$277,241	$277,241	$277,241	$277,241	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4	Site Maintenance			$1,760,177	10%	$1,760,177	$0	$0	$0	$0	$30,000	$30,000	$30,000	$99,414	$99,414	$99,414	$99,414	$99,414	$0	$0	$0	$0	$99,414	$0	$0	$0	$0	$99,414	$0	$0	$0	$0	$0	$0	$0	$0	$108,253	$0	$108,253
5	On-site Management			$5,728,488	34%	$5,728,488	$0	$0	$0	$0	$310,135	$310,135	$310,135	$606,140	$606,140	$606,140	$606,140	$606,140	$0	$0	$0	$0	$79,830	$0	$0	$0	$0	$79,830	$0	$0	$0	$0	$0	$0	$0	$0	$178,636	$0	$178,636
6	Transportation Costs			$994,297	6%	$994,297	$0	$0	$0	$0	$0	$0	$0	$64,628	$64,628	$64,628	$64,628	$64,628	$0	$0	$0	$0	$20,832	$0	$0	$0	$0	$20,832	$0	$0	$0	$0	$0	$0	$0	$0	$69,943	$0	$69,943
Post Closure Costs Subtotal				**$16,996,485**		$16,996,485	**$0**	**$0**	**$0**	**$0**	**$1,047,102**	**$1,047,102**	**$1,047,102**	**$1,789,405**	**$1,789,405**	**$1,789,405**	**$1,789,405**	**$1,789,405**	**$91,135**	**$91,135**	**$91,135**	**$91,135**	**$326,226**	**$91,135**	**$91,135**	**$91,135**	**$91,135**	**$326,226**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$391,847**	**$0**	**$391,847**

Other Add-on Costs

				Total	% of Total	Check Sum	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2128	2129
1	Contingency	12% % of above costs		$5,535,321		$5,535,321	$0	$0	$58,637	$68,830	$1,085,920	$1,408,920	$1,021,611	$227,224	$225,342	$227,224	$227,000	$236,300	$14,203	$14,203	$14,203	$14,203	$51,213	$14,203	$14,203	$14,203	$14,203	$51,213	$0	$0	$0	$0	$0	$0	$0	$0	$59,141	$0	$59,141
2	Worker's compensation	0% (Included in labour rates)		$0		$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3	Insurance	1.5% % of above labour costs		$199,509		$199,509	$0	$0	$2,113	$2,481	$39,140	$50,782	$36,822	$8,190	$8,122	$8,190	$8,182	$8,517	$512	$512	$512	$512	$1,846	$512	$512	$512	$512	$1,846	$0	$0	$0	$0	$0	$0	$0	$0	$2,132	$0	$2,132
4	Bonding	3% % of Implem. & C&M costs		$1,276,050		$1,276,050	$0.00	$0.00	$0.00	$0	$263,789	$344,539	$247,712	$54,017	$54,017	$54,017	$54,431	$56,286	$2,734	$2,734	$2,734	$2,734	$9,787	$2,734	$2,734	$2,734	$2,734	$9,787	$0	$0	$0	$0	$0	$0	$0	$0	$11,755	$0	$11,755
5	Contractor Profit	10% % of above costs		$4,612,767		$4,612,767	$0	$0	$48,865	$57,358	$904,933	$1,174,100	$851,343	$189,353	$187,785	$189,353	$189,167	$196,917	$11,835	$11,835	$11,835	$11,835	$42,677	$11,835	$11,835	$11,835	$11,835	$42,677	$0	$0	$0	$0	$0	$0	$0	$0	$49,284	$0	$49,284
6	Contract Administration	5% % of above costs		$2,306,384		$2,306,384	$0	$0	$24,432	$28,679	$452,467	$587,050	$425,671	$94,677	$93,892	$94,677	$94,584	$98,458	$5,918	$5,918	$5,918	$5,918	$21,339	$5,918	$5,918	$5,918	$5,918	$21,339	$0	$0	$0	$0	$0	$0	$0	$0	$24,642	$0	$24,642
Other Add-On Costs Subtotal				**$13,930,031**		$13,930,031	**$0**	**$0**	**$134,048**	**$157,349**	**$2,746,248**	**$3,565,390**	**$2,583,159**	**$573,460**	**$569,158**	**$573,460**	**$573,364**	**$596,477**	**$35,202**	**$35,202**	**$35,202**	**$35,202**	**$126,862**	**$35,202**	**$35,202**	**$35,202**	**$35,202**	**$126,862**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$146,954**	**$0**	**$146,954**

TOTAL			**$60,057,703**		**$60,057,703**	**$0**	**$0**	**$622,693**	**$730,933**	**$11,795,580**	**$15,306,387**	**$11,096,585**	**$2,466,991**	**$2,447,007**	**$2,466,991**	**$2,465,034**	**$2,565,643**	**$153,556**	**$153,556**	**$153,556**	**$153,556**	**$553,635**	**$153,556**	**$153,556**	**$153,556**	**$153,556**	**$553,635**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$639,794**	**$0**	**$639,794**

INFLATION FACTORED INTERIM CARE AND CLOSURE IMPLEMENTATION COSTS

	Area	Present Value	% of Total	Future Value	Year -4 2018	Year -3 2019	Year -2 2020	Year -1 2021	Year 1 2022	Year 2 2023	Year 3 2024	Year 4 2025	Year 5 2026	Year 6 2027	Year 7 2028	Year 8 2029	Year 9 2030	Year 10 2031	Year 11 2032	Year 12 2033	Year 13 2034	Year 14 2035	Year 15 2036	Year 16 2037	Year 17 2038	Year 18 2039	Year 19 2040	Year 20 2041	Year 21 2042	Year 22 2043	Year 23 2044	Year 24 2045	Year 25 2046	Year 26 2047	Year 27 2048	Year 28 2049	Year 107 2128	Year 108 2129
Direct Costs	Waste Dumps	$6,987,586	27%	$7,654,351	$0	$0	$0	$0	$3,025,435	$4,628,916	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Overburden Dumps	$395,276	2%	$445,145	$0	$0	$0	$0	$0	$0	$445,145	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Ore Stockpiles	$794,200	3%	$859,667	$0	$0	$0	$0	$859,667	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Open Pits	$620,582	2%	$692,025	$0	$0	$0	$0	$0	$342,587	$349,438	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Underground Openings	$81,637	0%	$88,367	$0	$0	$0	$0	$88,367	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	External Tailings Facilities	$57,291	0%	$62,634	$0	$0	$0	$0	$31,007	$31,627	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Roads	$136,050	1%	$153,214	$0	$0	$0	$0	$0	$0	$153,214	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Demolition	$1,976,253	8%	$2,184,210	$0	$0	$0	$0	$213,916	$1,636,457	$333,837	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Surface Infrastructure	$169,527	1%	$187,190	$0	$0	$0	$0	$45,875	$93,586	$47,729	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Water Detention Structures	$786,952	3%	$877,547	$0	$0	$0	$0	$0	$434,429	$443,118	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Yards/Laydown Areas	$758,907	3%	$842,166	$0	$0	$0	$0	$205,366	$209,474	$427,326	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Waste Disposal	$452,923	2%	$512,699	$0	$0	$0	$0	$0	$60,535	$382,549	$12,802	$13,058	$13,320	$30,434	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Surface Water Conveyance	$3,287,496	13%	$3,651,724	$0	$0	$0	$0	$711,698	$1,088,898	$1,851,127	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Water Treatment	$483,085	2%	$552,137	$0	$0	$0	$0	$0	$0	$466,161	$0	$0	$0	$0	$85,976	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Indirect	Mobilization-Demobilization	$324,077	1%	$359,944	$0	$0	$0	$0	$175,396	$0	$162,624	$0	$0	$0	$0	$21,924	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Transport Costs	$419,142	2%	$462,736	$0	$0	$0	$0	$154,256	$152,713	$155,767	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Road Maintenance	$791,662	3%	$874,001	$0	$0	$0	$0	$291,353	$288,440	$294,208	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	Construction Support	$5,263,751	21%	$5,811,219	$0	$0	$0	$0	$1,937,202	$1,917,830	$1,956,187	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
	QA and Pro. Mgmt.	$1,752,126	7%	$1,934,360	$0	$0	$0	$0	$644,830	$638,381	$651,149	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Subtotal - Closure Implementation Costs		**$25,538,523**		**$28,205,335**	**$0**	**$0**	**$0**	**$0**	**$8,384,368**	**$11,523,872**	**$8,119,580**	**$12,802**	**$13,058**	**$13,320**	**$30,434**	**$107,900**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

Monitoring Costs

			Present Value	% of Total	Future Value	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2128	2129
1	Planning and permitting	Reclamation Research and Planning	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Technical Studies and Investigations	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3		Engineering, Design, and Construction Plans	$254,816	7%	$281,375	$0	$0	$0	$0	$91,941	$93,779	$95,655	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4		Monitoring and Management Plans	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5		Permitting	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
1	Monitoring	Water Quality Monitoring	$391,312	11%	$432,097	$0	$0	$0	$0	$141,190	$144,014	$146,894	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Sediment Monitoring	$10,164	0%	$11,223	$0	$0	$0	$0	$3,667	$3,741	$3,815	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3		Biological Monitoring	$36,882	1%	$40,726	$0	$0	$0	$0	$13,307	$13,574	$13,845	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4		Geotechnical Monitoring	$37,227	1%	$41,107	$0	$0	$0	$0	$13,432	$13,701	$13,975	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5		Revegetation Monitoring	$31,590	1%	$34,883	$0	$0	$0	$0	$11,398	$11,626	$11,859	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
6		Annual Inspection Reporting	$7,116	0%	$7,858	$0	$0	$0	$0	$2,568	$2,619	$2,671	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Post Closure Costs Subtotal			**$769,108**		**$849,269**	**$0**	**$0**	**$0**	**$0**	**$277,503**	**$283,053**	**$288,714**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

Care and Maintenance Costs

			Present Value	% of Total	Future Value	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2128	2129
1	Water Treatment	Active System	$2,120,900	12%	$2,341,951	$0	$0	$0	$0	$765,243	$780,548	$796,159	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Passive System	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3	Reclamation Maintenance	(Cover/Reveg Repairs)	$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4	Site Maintenance		$90,000	1%	$99,380	$0	$0	$0	$0	$32,473	$33,122	$33,785	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5	On-site Management		$930,406	5%	$1,027,378	$0	$0	$0	$0	$335,701	$342,415	$349,263	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
6	Transportation Costs		$0	0%	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Post Closure Costs Subtotal			**$3,141,306**		$3,468,709	**$0**	**$0**	**$0**	**$0**	**$1,133,417**	**$1,156,085**	**$1,179,207**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

Other Add-on Costs

			Present Value	% of Total	Future Value	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2128	2129
1	Contingency	12% % of above costs	$3,516,451		$3,881,496	$0	$0	$0	$0	$1,175,435	$1,555,561	$1,150,500	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2	Worker's compensation	0% (Included in labour rates)	$0		$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3	Insurance	2% % of above labour costs	$126,743		$139,900	$0	$0	$0	$0	$42,366	$56,067	$41,467	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4	Bonding	3% % of Implem. & C&M costs	$856,039		$944,896	$0	$0	$0	$0	$285,534	$380,399	$278,964	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
5	Contractor Profit	10% % of above costs	$2,930,376		$3,234,580	$0	$0	$0	$0	$979,529	$1,296,301	$958,750	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
6	Contract Administration	5% % of above costs	$1,465,188		$1,617,290	$0	$0	$0	$0	$489,764	$648,150	$479,375	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
Other Add-On Costs Subtotal			**$8,894,796**		$9,818,162	**$0**	**$0**	**$0**	**$0**	**$2,972,627**	**$3,936,478**	**$2,909,056**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

TOTAL			**$38,343,734**		**$42,341,475**	**$0**	**$0**	**$0**	**$0**	**$12,767,915**	**$16,899,488**	**$12,496,557**	**$12,802**	**$13,058**	**$13,320**	**$30,434**	**$107,900**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**

NET PRESENT VALUE CALCULATIONS - LONG TERM MONITORING AND MAINTENANCE

Monitoring Costs

			Undiscounted	% of Total	NPV - 2018	Year -4 2018	Year -3 2019	Year -2 2020	Year -1 2021	Year 1 2022	Year 2 2023	Year 3 2024	Year 4 2025	Year 5 2026	Year 6 2027	Year 7 2028	Year 8 2029	Year 9 2030	Year 10 2031	Year 11 2032	Year 12 2033	Year 13 2034	Year 14 2035	Year 15 2036	Year 16 2037	Year 17 2038	Year 18 2039	Year 19 2040	Year 20 2041	Year 21 2042	Year 22 2043	Year 23 2044	Year 24 2045	Year 25 2046	Year 26 2047	Year 27 2048	Year 28 2049	Year 107 2128	Year 108 2129
1	Monitoring	Water Quality Monitoring	$781,111	22%	$440,095								$44,829	$43,608	$42,420	$41,265	$40,141	$20,444	$19,888	$19,346	$18,819	$18,306	$17,808	$17,323	$16,851	$16,392	$15,945	$0	$0	$0	$0	$0	$0	$0	$0	$0	$12,295	$0	$1,350
2		Sediment Monitoring	$47,432	1%	$18,286								$2,921	$0	$2,765	$0	$2,616	$0	$0	$0	$0	$2,279	$0	$0	$0	$0	$1,985	$0	$0	$0	$0	$0	$0	$0	$0	$1,506	$0	$165
3		Biological Monitoring	$172,117	5%	$66,354								$10,601	$0	$10,032	$0	$9,493	$0	$0	$0	$0	$8,268	$0	$0	$0	$0	$7,202	$0	$0	$0	$0	$0	$0	$0	$0	$5,464	$0	$600
4		Geotechnical Monitoring	$317,426	9%	$119,080								$10,700	$10,409	$10,125	$9,850	$9,581	$0	$0	$0	$0	$15,614	$0	$0	$0	$0	$13,600	$0	$0	$0	$0	$0	$0	$0	$0	$10,319	$0	$1,133
5		Revegetation Monitoring	$168,481	5%	$74,024								$9,080	$8,833	$8,592	$8,358	$8,131	$0	$0	$0	$0	$7,082	$0	$0	$0	$0	$6,169	$0	$0	$0	$0	$0	$0	$0	$0	$4,680	$0	$514
6		Annual Inspection Reporting	$274,760	8%	$80,114								$2,045	$1,990	$1,935	$1,883	$1,831	$0	$0	$0	$0	$16,074	$0	$0	$0	$0	$14,001	$0	$0	$0	$0	$0	$0	$0	$0	$10,622	$0	$1,166
Post Closure Costs Subtotal			**$1,761,326**		**$797,952**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$80,177**	**$64,839**	**$75,869**	**$61,355**	**$71,793**	**$20,444**	**$19,888**	**$19,346**	**$18,819**	**$67,623**	**$17,808**	**$17,323**	**$16,851**	**$16,392**	**$58,902**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$44,886**	**$0**	**$4,928**

Care and Maintenance Costs

			Undiscounted	% of Total	NPV - 2018	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2128	2129
1	Water Treatment	Active System	$4,446,180	26%	$3,492,933								$609,619	$593,015	$576,862	$561,150	$545,866	$68,451	$66,586	$64,773	$63,008	$61,292	$59,623	$57,999	$56,419	$54,882	$53,388	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
2		Passive System	$560,239	3%	$246,147								$30,193	$29,371	$28,571	$27,793	$27,036	$0	$0	$0	$0	$23,549	$0	$0	$0	$0	$20,512	$0	$0	$0	$0	$0	$0	$0	$0	$15,562	$0	$1,709
3	Reclamation Maintenance	(Cover/Reveg Repairs)	$1,386,205	8%	$1,131,962								$239,066	$232,554	$226,220	$220,058	$214,064	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
4	Site Maintenance		$1,670,177	10%	$713,783								$85,725	$83,390	$81,119	$78,909	$76,760	$0	$0	$0	$0	$66,860	$0	$0	$0	$0	$58,238	$0	$0	$0	$0	$0	$0	$0	$0	$48,113	$0	$5,282
5	On-site Management		$4,798,081	28%	$2,876,918								$522,676	$508,440	$494,591	$481,120	$468,016	$0	$0	$0	$0	$53,689	$0	$0	$0	$0	$46,765	$0	$0	$0	$0	$0	$0	$0	$0	$79,395	$0	$8,717
6	Transportation Costs		$994,297	6%	$408,186								$55,729	$54,211	$52,735	$51,298	$49,901	$0	$0	$0	$0	$14,011	$0	$0	$0	$0	$12,204	$0	$0	$0	$0	$0	$0	$0	$0	$31,086	$0	$3,413
Post Closure Costs Subtotal			**$13,855,178**		$8,869,929	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$1,543,008**	**$1,500,981**	**$1,460,098**	**$1,420,329**	**$1,381,643**	**$68,451**	**$66,586**	**$64,773**	**$63,008**	**$219,402**	**$59,623**	**$57,999**	**$56,419**	**$54,882**	**$191,106**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$174,157**	**$0**	**$19,120**

Other Add-on Costs

			Undiscounted	% of Total	NPV - 2018	2018	2019	2020	2021	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035	2036	2037	2038	2039	2040	2041	2042	2043	2044	2045	2046	2047	2048	2049	2128	2129
1	Contingency	12% % of above costs	$1,891,402		$1,173,931								$195,936	$189,020	$185,407	$180,180	$182,453	$10,667	$10,377	$10,094	$9,819	$34,443	$9,292	$9,039	$8,792	$8,553	$30,001	$0	$0	$0	$0	$0	$0	$0	$0	$26,285	$0	$2,886
2	Worker's compensation	0% (Included in labour rates)	$0		$0								$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0	$0
3	Insurance	2% % of above labour costs	$68,172		$42,312								$7,062	$6,813	$6,683	$6,494	$6,576	$384	$374	$364	$354	$1,241	$335	$326	$317	$308	$1,081	$0	$0	$0	$0	$0	$0	$0	$0	$947	$0	$104
4	Bonding	3% % of Implem. & C&M costs	$420,011		$269,544								$46,579	$45,310	$44,076	$43,204	$43,459	$2,054	$1,998	$1,943	$1,890	$6,582	$1,789	$1,740	$1,693	$1,646	$5,733	$0	$0	$0	$0	$0	$0	$0	$0	$5,225	$0	$574
5	Contractor Profit	10% % of above costs	$1,576,169		$978,276								$163,280	$157,517	$154,506	$150,150	$152,044	$8,890	$8,647	$8,412	$8,183	$28,702	$7,743	$7,532	$7,327	$7,127	$25,001	$0	$0	$0	$0	$0	$0	$0	$0	$21,904	$0	$2,405
6	Contract Administration	5% % of above costs	$788,084		$489,138								$81,640	$78,758	$77,253	$75,075	$76,022	$4,445	$4,324	$4,206	$4,091	$14,351	$3,872	$3,766	$3,663	$3,564	$12,500	$0	$0	$0	$0	$0	$0	$0	$0	$10,952	$0	$1,202
Other Add-On Costs Subtotal			**$4,743,838**		$2,953,201	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$494,496**	**$477,418**	**$467,925**	**$455,104**	**$460,555**	**$26,440**	**$25,720**	**$25,019**	**$24,338**	**$85,320**	**$23,030**	**$22,403**	**$21,792**	**$21,199**	**$74,317**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$65,314**	**$0**	**$7,171**

TOTAL			**$20,360,342**		**$12,621,082**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$2,117,681**	**$2,043,238**	**$2,003,892**	**$1,936,788**	**$1,913,990**	**$115,335**	**$112,193**	**$109,138**	**$106,165**	**$372,345**	**$100,460**	**$97,724**	**$95,062**	**$92,473**	**$324,325**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$284,358**	**$0**	**$31,219**

GRAND-TOTAL					**$54,962,557**	**$0**	**$0**	**$0**	**$0**	**$12,767,915**	**$16,899,488**	**$12,496,557**	**$2,130,484**	**$2,056,297**	**$2,017,212**	**$1,967,223**	**$2,021,890**	**$115,335**	**$112,193**	**$109,138**	**$106,165**	**$372,345**	**$100,460**	**$97,724**	**$95,062**	**$92,473**	**$324,325**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$0**	**$284,358**	**$0**	**$31,219**

Worksheet 13 - End-of-Mine Liability Estimate - Annual C&M Costs

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

Summary of Annual Care and Maintenance Costs

		Annual Cost Per Phase						Number of Years				
		1 - Interim	2 - Active	3 - PC 1	4 - PC 2	5 - Perpetual	n/a	Total Cost	1 - Interim	2 - Active	3 - PC 1	4 - PC 2	5 - Perpetual
1	Active Water Treatment System		$706,967	$706,967	$91,135	$0	$0	$6,567,079	0	3	5	10	0
2	Passive Treatment System		$0	$35,015	$35,015	$35,015	$0	$560,239	0	3	5	2	9
3	Reclamation Maintenance		$0	$277,241	$0	$0	$0	$1,386,205	0	3	5	2	9
4	Site Maintenance		$30,000	$99,414	$99,414	$108,253	$0	$1,760,177	0	3	5	2	9
5	On-site Management		$310,135	$606,140	$79,830	$178,636	$0	$5,728,488	0	3	5	2	9
6	Transportation Costs		$0	$64,628	$20,832	$69,943	$0	$994,297	0	3	5	2	9

NOTES:
1. This table is used as a check to the NPV calculation spreadsheet.

Phase 1 - Annual Care & Maintenance Costs - Prior to Closure

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S1.1	**Water Treatment**															**$706,967**	
S1.1.1	*Active Treatment System*														$706,967		
S1.1.1.1		Operation and maintenance		1	yr						$511,807	$65,194	$38,831	$615,832			See details on 'Water Treatment' worksheet.
S1.1.1.2		Capital Replacement	annual allowance	1	yr						$86,932	$4,203	$0	$91,135			See details on 'Water Treatment' worksheet.
S1.2	**Onsite Management**															**$180,419**	
S1.2.1	*Field Support Staff*														$0		
S1.2.1.1		Mine Manager		0	months	360	site working hours per month			$23,408				$0			(Notes: staffing costs include cross-shifts)
S1.2.1.2		Office/Camp manager		0	months	360	site working hours per month			$22,267				$0			
S1.2.1.3		Administrative Assistants/HR/Accounting		0	months	360	site working hours per month			$16,600				$0			
S1.2.1.4		Water Treatment Staff Operator		0	months	360	site working hours per month			$18,912				$0			
S1.2.1.5		Environmental/Safety Manager		0	months	360	site working hours per month			$22,267				$0			
S1.2.1.6		Environmental Technicians		0	months	360	site working hours per month			$16,046				$0			
S1.2.1.7		Mechanic		0	months	360	site working hours per month			$17,769				$0			
S1.2.1.8		Tradesmen		0	months	360	site working hours per month			$18,912				$0			
S1.2.1.9		General Labour/helpers		0	months	360	site working hours per month			$12,793				$0			
S1.2.2	*Field support Vehicles*														$107,579		
S1.2.2.1		Pick-up trucks (3 required)		12	months					$4,446				$53,350			
S1.2.2.2		Mechanic service vehicle		0	months					$11,803				$0			
S1.2.2.3		Emergency transport vehicle		12	months					$4,519				$54,229			
S1.2.3	*Field Support Equipment/Supplies*														$72,840		
S1.2.3.1		Office supplies		12	months					$1,000				$12,000			
S1.2.3.2		Communications		12	months					$1,000				$12,000			
S1.2.3.3		Misc. supplies		12	months					$500				$6,000			
S1.2.3		Camp Operation		0	man-days					$80				$0			
S1.2.3		Power, heat, fuel		12	months					$3,570				$42,840			
S1.3	**Site Maintenance**														$359,187	**$359,187**	
S1.3.1		Equipment Maintenance		1	ls		$20,000							$20,000			Labour included in field support staff
S1.3.2		Road Maintenance	Grader (assume 80 hrs/month)	1	ls	960		$44	$106	$150	$0	$42,612	$101,443	$144,056			Operator cost included in Field support staff
S1.3.3		Earthwork Repair allowance (assume 40 hrs per month)		1	ls	480	$0	$132	$344	$344			$165,132	$165,132			Task Code C.2.14
S1.3.4		Sundry equipment/consumables allowance		1	ls					$30,000				$30,000			Allowance for pump maintenance/fuel etc.
S1.4	**Transportation Costs**														$0	**$0**	
S1.4.1		Barge Operations		0	months					$10,832				$0			
S1.4.2		Staffing Bus trips during barge operation period		0	ea					$678				$0			One per week
S1.4.3		Air transport and airstrip operations		0	flights					$3,000				$0			One flight per week
S1.4.4		Helicopter transport		0	flights					$3,000				$0			One flight per week
TOTAL														**$1,246,573**	**$1,246,573**	**$1,246,573**	

Phase 2 - Annual Care & Maintenance Costs During Active Closure

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S2.1	**Water Treatment**															**$706,967**	
S2.1.1	*Active Treatment System*														$706,967		
S2.1.1.1		Operation and maintenance		1	yr						$511,807	$65,194	$38,831	$615,832			See details on 'Water Treatment' worksheet.
S2.1.1.2		Capital Replacement	annual allowance	1	yr						$86,932	$4,203	$0	$91,135			See details on 'Water Treatment' worksheet.
S2.2	**Onsite Management**															**$310,135**	
S2.2.1	*Field Support Staff*														$247,815		
S2.2.1.1		Water Treatment Staff Operator		5	months	360	site working hours per month			$18,912				$94,560			
S2.2.1.2		Environmental/Safety Manager		4	months	360	site working hours per month			$22,267				$89,069			
S2.2.1.3		Environmental Technicians		4	months	360	site working hours per month			$16,046				$64,185			
S2.2.2	*Field support Vehicles*														$31,121		
S2.2.2.1		Pick-up trucks (1 required)		7	months					$4,446				$31,121			
S2.2.3	*Field Support Equipment/Supplies*														$31,200		
S2.2.3.1		Camp Operation		390	man-days					$80				$31,200			
S2.3	**Site Maintenance**														$30,000	**$30,000**	
S2.3.1		Pumping equipment/consumables allowance		1	ls					$30,000				$30,000			Allowance for pump maintenance/fuel etc.
S2.4	**Transportation Costs**														$0	**$0**	
S2.4.1		Barge Operations		0	months					$10,832				$0			Included in construction indirect costs
TOTAL														**$1,047,102**	**$1,047,102**	**$1,047,102**	

Phase 3 - Annual Care & Maintenance Costs - Post Closure I

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours or Cost Code Ref.	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S3.1	**Water Treatment**															**$741,982**	
S3.1.1	*Active Treatment System*														$706,967		
S3.1.1.1		Operation and maintenance		1	yr						$511,807	$65,194	$38,831	$615,832			See details on 'Water Treatment' worksheet.
S3.1.1.2		Capital Replacement	annual allowance	1	yr						$86,932	$4,203	$0	$91,135			See details on 'Water Treatment' worksheet.
S3.1.1	*Passive Treatment System*														$35,015		
S3.1.1.1		Operation and maintenance		1	yr						$10,000	$11,986	$13,029	$35,015			See details on 'Water Treatment' worksheet.
S3.2	**Onsite Management**															**$606,140**	
S3.2.1	*Field Support Staff*														$377,100		
S3.2.1.1		Mine Manager		3	months	360	site working hours per month			$23,408				$58,521			(Notes: staffing costs include cross-shifts)
S3.2.1.2		Office/Camp manager		3	months	360	site working hours per month			$22,267				$66,802			
S3.2.1.3		Administrative Assistants/HR/Accounting		1	months	360	site working hours per month			$16,600				$20,749			
S3.2.1.4		Water Treatment Staff Operator		2	months	360	site working hours per month			$18,912				$33,096			
S3.2.1.5		Environmental/Safety Manager		3	months	360	site working hours per month			$22,267				$55,668			
S3.2.1.6		Environmental Technicians		3	months	360	site working hours per month			$16,046				$40,116			
S3.2.1.7		Mechanic		2	months	360	site working hours per month			$17,769				$35,538			
S3.2.1.8		Tradesmen		2	months	360	site working hours per month			$18,912				$37,824			
S3.2.1.9		General Labour/helpers		2	months	360	site working hours per month			$12,793				$28,785			
S3.2.2	*Field support Vehicles*														$112,890		
S3.2.2.1		Pick-up trucks (3 required)		15	months					$4,446				$66,688			
S3.2.2.2		Mechanic service vehicle		2	months					$11,803				$23,607			
S3.2.2.3		Emergency transport vehicle		5	months					$4,519				$22,596			
S3.2.3	*Field Support Equipment/Supplies*														$116,150		
S3.2.3.1		Office supplies		5	months					$1,000				$5,000			
S3.2.3.2		Communications		5	months					$1,000				$5,000			
S3.2.3.3		Misc. supplies		5	months					$500				$2,500			
S3.2.3		Camp Operation		1,073	man-days					$80				$85,800			
S3.2.3		Power, heat, fuel		5	months					$3,570				$17,850			
S3.3	**Reclamation Maintenance**															**$277,241**	
S3.3.1	*Cover repairs*														$249,419		
S3.3.1.1		Cover Repair allowance	Load, haul, dump, spread (spoil lo	1%	of total	1,336,364	m3 of cover required replacement at:			$5.31	/m3			$71,022			Assumes 5% of area reseeded over 5 yr PC1 period.
S3.3.1.2			Seed/Fertilize: broadcast seeding	67	ha					$2,669.89	/ha			$178,397			
S3.3.2.1	*Revegetation*														$27,822		
S3.3.2.2		Reseeding allowance	Seed/Fertilize: broadcast seeding	3%	of total	347	ha requires reseeding at			$2,669.89	/ha			$27,822			Assumes 15% of area reseeded over 5 yr PC1 perio
S3.3	**Site Maintenance**														$99,414	**$99,414**	
S3.3.1		Equipment Maintenance		1	ls		$20,000							$20,000			Labour included in field support staff
S3.3.2		Road Maintenance	Grader (assume 40 hrs/month)	1	ls	200		$44	$106	$150	$0	$8,878	$21,134	$30,012			Operator cost included in Field support staff
S3.3.3		Earthwork Repair allowance (assume 20 hrs per month)		1	ls	100	$0	$132	$344	$344			$34,402	$34,402			Task Code C.2.14
S3.3.4		Sundry equipment/consumables allowance		1	ls					$15,000				$15,000			Allowance for pump maintenance/fuel etc.
S3.4	**Transportation Costs**														$64,628	**$64,628**	
S3.4.1		Barge Operations		3	months					$10,832				$32,497			
S3.4.2		Staffing Bus trips during barge operation period		12	ea					$678				$8,131			One per week
S3.4.3		Air transport and airstrip operations		8	flights					$3,000				$24,000			One flights per week
TOTAL														**$1,789,405**	**$1,789,405**	**$1,789,405**	

Phase 4 - Annual Care & Maintenance Costs - Post Closure II

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours or Cost Code Ref.	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S4.1	**Water Treatment**															**$126,150**	
S4.1.1	*Active Treatment System*														$91,135		
S4.1.1.2		Capital Replacement	annual allowance	1	yr						$86,932	$4,203	$0	$91,135			See details on 'Water Treatment' worksheet.
S4.1.1	*Passive Treatment System*														$35,015		
S4.1.1.1		Operation and maintenance		1	yr						$10,000	$11,986	$13,029	$35,015			See details on 'Water Treatment' worksheet.
S4.2	**Onsite Management**															**$79,830**	
S4.2.1	*Field Support Staff*														$44,847		
S4.2.1.1		Project Manager		0.5	months	360	site working hours per month			$23,408				$11,704			(Notes: staffing costs include cross-shifts)
S4.2.1.2		Passive Treatment specialist		0.5	months	360	site working hours per month			$42,120				$21,060			
S4.2.1.3		Mechanic		0.5	months	360	site working hours per month			$17,769				$8,885			
S4.2.1.4		General Labour/helpers		0.3	months	360	site working hours per month			$12,793				$3,198			
S4.2.2	*Field support Vehicles*														$19,313		
S4.2.2.1		Pick-up trucks (2 required)		2.0	months					$4,446				$8,892			
S4.2.2.2		Mechanic service vehicle		0.5	months					$11,803				$5,902			
S4.2.2.3		Emergency transport vehicle		1.0	months					$4,519				$4,519			
S4.2.3	*Field Support Equipment/Supplies*														$15,670		
S4.2.3.1		Office supplies		1	months					$1,000				$1,000			
S4.2.3.2		Communications		1	months					$1,000				$1,000			
S4.2.3.3		Misc. supplies		1	months					$500				$500			
S4.2.3		Camp Operation		120	man-days					$80				$9,600			
S4.2.3		Power, heat, fuel		1	months					$3,570				$3,570			
S4.3	**Reclamation Maintenance**															**$0**	
S4.3.1	*Cover repairs*														$0		
S4.3.2	*Revegetation*														$0		
S4.3	**Site Maintenance**														$29,882	**$29,882**	
S4.3.1		Equipment Maintenance (parts, supplies)		1	ls		$10,000							$10,000			Labour included in field support staff
S4.3.2		Road Maintenance	Grader (assume 20 hrs/month)	1	ls	20		$44	$106	$150	$0	$888	$2,113	$3,001			Operator cost included in Field support staff
S4.3.3		Earthwork Repair allowance (assume 20 hrs per month)		1	ls	20	$0	$132	$344	$344			$6,880	$6,880			Task Code C.2.14
S4.3.4		Sundry equipment/consumables allowance		1	ls					$10,000				$10,000			Allowance for pump maintenance/fuel etc.
S4.4	**Transportation Costs**														$20,832	**$20,832**	
S4.4.1		Barge Operations		1	months					$10,832				$10,832			
S4.4.2		Staffing Bus trips during barge operation period		0	ea					$678				$0			Pick-up trucks used (time included in labour rate)
S4.4.3		Barge mob/demob		1	LS					$10,000				$10,000			
TOTAL														**$256,693**	**$256,693**	**$256,693**	

Phase 5 - Perpetual Care & Maintenance Event Year Annual Costs

							Unit Rates				Activity Totals				Subtotals		
WBS	Facility/Area	Task	Activity	Qnty	Unit	Hours or Cost Code Ref.	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Total Unit Rate ($/unit)	Material Cost	Labour Cost	Equipment Cost	Cost	WBS Level 2	WBS Level 1	Source / Comments
S5.1	**Water Treatment**															**$35,015**	
S5.1.1	*Passive Treatment System*														$35,015		
S5.1.1.1		Operation and maintenance		1	yr						$10,000	$11,986	$13,029	$35,015			See details on 'Water Treatment' worksheet.
S5.2	**Onsite Management**															**$178,636**	
S5.2.1	*Field Support Staff*														$129,326		
S5.2.1.1		Project Manager		1.0	months	360	site working hours per month			$23,408				$23,408			(Notes: staffing costs include cross-shifts)
S5.2.1.2		Passive Treatment specialist		1.0	months	360	site working hours per month			$42,120				$42,120			
S5.2.1.3		Site Engineer		1.0	months	360	site working hours per month			$42,120				$42,120			
S5.2.1.4		HD mechanic		0.5	months	360	site working hours per month			$17,769				$8,885			
S5.2.1.5		General Labour/helpers		1.0	months	360	site working hours per month			$12,793				$12,793			
S5.2.2	*Field support Vehicles*														$19,239		
S5.2.2.1		Pick-up trucks (3 required)		3	months					$4,446				$13,338			
S5.2.2.2		Mechanic service vehicle		1	months					$11,803				$5,902			
S5.2.3	*Field Support Equipment/Supplies*														$30,070		
S5.2.3.1		Office supplies		1	months					$1,000				$1,000			
S5.2.3.2		Communications		1	months					$1,000				$1,000			
S5.2.3.3		Misc. supplies		1	months					$500				$500			
S5.2.3.4		Camp Operation		300	man-days					$80				$24,000			
S5.2.3.5		Power, heat, fuel		1	months					$3,570				$3,570			
S5.3	**Reclamation Maintenance**															**$0**	
S5.4	**Site Maintenance**														$108,253	**$108,253**	
S5.4.1		Grader Maintenance	Grader (assume 2 weeks, 10hrs/d	1	ls	140		$44	$106	$150	$0	$6,214	$14,794	$21,008			Operator cost included in Field support staff
S5.4.2		Earthwork Repair allowance (assume 3 weeks, 10hrs/day)		1	ls	210	$0	$132	$344	$344			$72,245	$72,245			Task Code C.2.14
S5.4.3		Big Creek Bridge - Capital replacement allowance		1	ls					$15,000				$15,000			Allowance for supplies, pumping, etc.
S5.4.4		Sundry allowance		1	ls					$2,500				$2,500			Allowance for supplies, pumping, etc.
S5.5	**Transportation Costs**														$69,943	**$69,943**	
S5.5.1		Barge Operations		1	months					$10,832				$10,832			
S5.5.2		Barge mobilization/demob and set up		1	ls					$10,000				$10,000			assumed
S5.5.3		Camp Mob/demob		1	ls					$10,000				$10,000			assumed
S5.5.4		Equipment mobilization and demobilization		2	ls					$19,555				$39,110.91			See details on 'mob-demob 'worksheet
TOTAL														**$394,347**	**$391,847**	**$391,847**	

Worksheet 14 - End-of-Mine Liability Estimate - Implementation Costs

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
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							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	EOM Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
DIRECT COSTS																					
A1	**Waste Dumps**																			**$6,987,586**	
A1.1	*Main Waste Dump (inc. MWDE)*																		$121,985		
A1.1.1.1		Regrade	Flat areas: Re-grade to form tertiary drainage catchments	0	hrs	C.2.13	0	$44.39	$0	$194.68	$0	$67.01	0	$0	$0.00	$0	$306.07	$0			
A1.1.1.2			Slopes: Re-grade to 3H:1V	0	hrs	C.2.13	0	$44.39	$0	$194.68	$0	$67.01	0	$0	$0.00	$0	$306.07	$0			
A1.1.2.1		Cover	Flat area cover material: Load, haul, dump spread	0	Cm3	R.050	0.0	$0.82	$0	$2.67	$0	$0.88	0	$0	$0.00	$0	$4.37	$0			
A1.1.2.2			Slope area cover material: Load, haul, dump along crest	0	Cm3	R.051	0.0	$0.72	$0	$2.37	$0	$0.80	0	$0	$0.00	$0	$3.89	$0			
A1.1.2.3			Slope area cover material: Spread down slope	0	hrs	C.2.13	0.0	$44.39	$0	$194.68	$0	$67.01	0	$0	$0.00	$0	$306.07	$0			
A1.1.3.1		Revegetate	Seed/Fertilize: tractor application	0	ha	C.6.01	0.0	$1,029.50	$0	$401.72	$0	$232.94	0	$0	$1,005.72	$0	$2,670	$0			
A1.1.3.2			Tree seedling application (1,000 stems/ha)	52	ha	C.6.06	2,668.2	$1,832.33	$95,397	$25.98	$1,353	$15.00	813	$781	$469.71	$24,455	$2,343	$121,985			
A1.2	*Main Waste Dump Wrap*																		$350,260		
A1.2.1.1		Regrade	Flat areas: Re-grade to form tertiary drainage catchments	6	hrs	C.2.13	6	$44.39	$278	$194.68	$1,218	$67.01	437	$419	$0.00	$0	$306.07	$1,914			
A1.2.1.2			Slopes: Re-grade to 3H:1V	255	hrs	C.2.13	255	$44.39	$11,334	$194.68	$49,709	$67.01	17,822	$17,110	$0.00	$0	$306.07	$78,152			
A1.2.2.1		Cover	Flat area cover material: Load, haul, dump spread	8,223	Cm3	R.050	153.4	$0.82	$6,757	$2.67	$21,957	$0.88	7,527	$7,226	$0.00	$0	$4.37	$35,939			
A1.2.2.2			Slope area cover material: Load, haul, dump along crest	42,447	Cm3	R.051	692.9	$0.72	$30,484	$2.37	$100,640	$0.80	35,215	$33,806	$0.00	$0	$3.89	$164,930			
A1.2.2.3			Slope area cover material: Spread down slope	61	hrs	C.2.13	60.5	$44.39	$2,686	$194.68	$11,781	$67.01	4,224	$4,055	$0.00	$0	$306.07	$18,523			
A1.2.3.1		Revegetate	Seed/Fertilize: tractor application	10	ha	C.6.01	335.7	$1,029.50	$10,433	$401.72	$4,071	$232.94	2,459	$2,361	$1,005.72	$10,192	$2,670	$27,057			
A1.2.3.2			Tree seedling application (1,000 stems/ha)	10	ha	C.6.06	519.4	$1,832.33	$18,569	$25.98	$263	$15.00	158	$152	$469.71	$4,760	$2,343	$23,744			
A1.3	*Southwest Dump (excluding high grade waste area)*																		$147,053		
A1.3.1		Re-grade	Slopes: Re-grade remaining to 3H:1V	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A1.3.2.1		Cover	Load, haul, dump spread	0	m3	R.059	0.0	$0.77	$0.00	$2.49	$0	$0.82	0	$0.00	$0.00	$0.00	$4.08	$0			
A1.3.3.1		Revegetate	Seed/Fertilize: tractor application	0.00	ha	C.6.01	0.0	$1,029.50	$0.00	$401.72	$0	$232.94	0	$0.00	$1,005.72	$0.00	$2,669.89	$0			
A1.3.3.2			Tree seedling application (1,000 stems/ha)	62.76	ha	C.6.06	3,216.6	$1,832.33	$115,000.98	$25.98	$1,631	$15.00	981	$941.43	$469.71	$29,479.97	$2,343.03	$147,053			
A1.4	*Southwest Dump High-Grade Waste*																		$249,632		
A1.4.1		Relocate High-grade waste	Load, haul, dump, spread in pit (below long-term water elevation)	45,711	Bm3	R.069	688.0	$0.66	$30,283.89	$2.19	$100,276	$0.73	34,965	$33,566.76	$0.00	$0.00	$3.59	$164,127			
A1.4.2		Regrade	Flat areas: Re-grade to form tertiary drainage catchments	31	hrs	C.2.13	31.1	$44.39	$1,378.38	$194.68	$6,045	$67.01	2,168	$2,080.81	$0.00	$0.00	$306.07	$9,505			
A1.4.3		Cover	Flat area cover material: Load, haul, dump spread	15,583	Cm3	R.060	257.8	$0.73	$11,362.15	$2.37	$36,935	$0.78	12,585	$12,081.20	$0.00	$0.00	$3.87	$60,378			
A1.4.4.1		Revegetate	Seed/Fertilize: broadcast seeding	3.12	ha	C.6.01	103.2	$1,029.50	$3,208.44	$401.72	$1,252	$232.94	756	$725.97	$1,005.72	$3,134.33	$2,669.89	$8,321			
A1.4.4.2			Tree seedling application (1,000 stems/ha)	3.12	ha	C.6.06	159.7	$1,832.33	$5,710.47	$25.98	$81	$15.00	49	$46.75	$469.71	$1,463.85	$2,343.03	$7,302			
A1.5	*Main Pit SAT*																		$4,103,230		
A1.5.1.1		Relocate SAT below water	Load, haul, dump, spread in pit	1,040,000	Cm3	R.012	17,781.5	$0.75	$785,076.51	$1.86	$1,935,773	$0.51	555,290	$533,078.26	$0.00	$0.00	$3.13	$3,253,928			
A1.5.1.2			Allowance for additional SAT material from SAT Dump	30,000	Cm3	R.012	512.9	$0.75	$22,646.44	$1.86	$55,840	$0.51	16,018	$15,377.26	$0.00	$0.00	$3.13	$93,863			
A1.5.2		Backfill SAT Excavation	Load, haul, dump WR to form buttress below MPD	245,000	Cm3	R.013	3,226.9	$0.58	$142,343.39	$1.89	$462,905	$0.61	156,449	$150,191.13	$0.00	$0.00	$3.08	$755,439			Cover and revegetation of backfill included in MPD (A1.6)
A1.6	*Main Pit Dump (Incl. SWB, In-Pit Dumps)*																		$686,521		
A1.6.2.1		Regrade	Flat areas: Re-grade to form tertiary drainage catchments	65	hrs	C.2.13	65.1	$44.39	$2,891.84	$194.68	$12,683	$67.01	4,547	$4,365.54	$0.00	$0.00	$306.07	$19,941			
A1.6.2.2			Slopes: Re-grade to 3H:1V	207	hrs	C.2.13	206.7	$44.39	$9,174.86	$194.68	$40,240	$67.01	14,428	$13,850.42	$0.00	$0.00	$306.07	$63,265			
A1.6.3.1		Cover	Flat area cover material: Load, haul, dump spread	45,270	m3	R.053	732.0	$0.71	$32,263.18	$2.32	$104,877	$0.76	35,734	$34,304.93	$0.00	$0.00	$3.79	$171,445			
A1.6.3.2			Slope area cover material: Load, haul, dump along crest	84,053	m3	R.054	1,165.0	$0.61	$51,284.32	$2.02	$169,813	$0.68	59,212	$56,843.70	$0.00	$0.00	$3.31	$277,941			
A1.6.3.3			Slope area cover material: Spread down slope	79	hrs	C.2.13	79.3	$44.39	$3,520.01	$194.68	$15,438	$67.01	5,535	$5,313.82	$0.00	$0.00	$306.07	$24,272			
A1.6.4.1		Revegetate	Seed/Fertilize: broadcast seeding	25.9	ha	C.6.01	856.8	$1,029.50	$26,627.72	$401.72	$10,390	$232.94	6,276	$6,025.02	$1,005.72	$26,012.60	$2,669.89	$69,056			
A1.6.4.2			Tree seedling application (1,000 stems/ha)	25.9	ha	C.6.06	1,325.6	$1,832.33	$47,392.66	$25.98	$672	$15.00	404	$387.97	$469.71	$12,148.89	$2,343.03	$60,602			
A1.7.0.0	*Mill Valley Fill Extension (Stage 1 and 2)*																		$40,934		
A1.7.1.1		Complete Fill Placement	Waste Rock fill: Load, haul, dump spread	0	m3	R.041	0.0	$0.97	$0.00	$3.15	$0	$1.04	0	$0.00	$0.00	$0.00	$5.15	$0			
A1.7.2.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A1.7.2.2			Slopes: Re-grade to 3H:1V	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A1.7.3.1		Cover	Flat area cover material: Load, haul, dump spread	0	m3	R.056	0.0	$0.90	$0.00	$2.91	$0	$0.96	0	$0.00	$0.00	$0.00	$4.77	$0			
A1.7.3.2			Slope area cover material: Load, haul, dump along crest	0	m3	R.057	0.0	$0.78	$0.00	$2.59	$0	$0.87	0	$0.00	$0.00	$0.00	$4.24	$0			
A1.7.3.3			Slope area cover material: Spread down slope	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A1.7.4.1		Revegetate	Seed/Fertilize: broadcast seeding	0.00	ha	C.6.01	0.0	$1,029.50	$0.00	$401.72	$0	$232.94	0	$0.00	$1,005.72	$0.00	$2,669.89	$0			
A1.7.4.2			Tree seedling application (1,000 stems/ha)	17.47	ha	C.6.06	895.4	$1,832.33	$32,011.85	$25.98	$454	$15.00	273	$262.06	$469.71	$8,206.09	$2,343.03	$40,934			
A1.8.0.0	*Ridgetop Waste Dump*																		$1,287,970		
A1.8.1.1		Regrade	Flat areas: Re-grade to form tertiary drainage catchments	88	hrs	C.2.13	88	$44.39	$3,903	$194.68	$17,119	$67.01	6,138	$5,892	$0.00	$0	$306.07	$26,915			
A1.8.1.2			Slopes: Re-grade to 3H:1V	383	hrs	C.2.13	383	$44.39	$17,021	$194.68	$74,651	$67.01	26,765	$25,695	$0.00	$0	$306.07	$117,367			
A1.8.2.1		Cover	Flat area cover material: Load, haul, dump spread	22,063	Cm3	R.063	590.5	$1.18	$26,007	$3.83	$84,514	$1.26	28,971	$27,812	$0.00	$0	$6.27	$138,332			
A1.8.2.2			Slope area cover material: Load, haul, dump along crest	142,998	Cm3	R.064	3,348.7	$1.03	$147,324	$3.40	$486,381	$1.14	170,188	$163,381	$0.00	$0	$5.57	$797,086			
A1.8.2.3			Slope area cover material: Spread down slope	140	hrs	C.2.13	139.8	$44.39	$6,205	$194.68	$27,213	$67.01	9,757	$9,367	$0.00	$0	$306.07	$42,784			
A1.8.3.1		Revegetate	Seed/Fertilize: tractor application	33	ha	C.6.01	1,093.6	$1,029.50	$33,986	$401.72	$13,262	$232.94	8,010	$7,690	$1,005.72	$33,201	$2,670	$88,139			
A1.8.3.2			Tree seedling application (1,000 stems/ha)	33	ha	C.6.06	1,691.9	$1,832.33	$60,489	$25.98	$858	$15.00	516	$495	$469.71	$15,506	$2,343	$77,348			
A2	**Overburden Dumps**																			**$395,276**	
A2.1	*Area 118 Pit Backfill Dump*																		$16,268		
A2.1.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A2.1.1.2			Berm: Regrade to 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A2.1.2.1		Revegetate	Seed/Fertilize: broadcast seeding	0	ha	C.6.01	0.0	$1,029.50	$0.00	$401.72	$0	$232.94	0	$0.00	$1,005.72	$0.00	$2,669.89	$0			
A2.1.2.2			Tree seedling application (1,000 stems/ha)	7	ha	C.6.06	355.8	$1,832.33	$12,722.54	$25.98	$180	$15.00	108	$104.15	$469.71	$3,261.36	$2,343.03	$16,268			
A2.2	*Ice-rich Overburden Dump*																		$13,827		
A2.2.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A2.2.1.2			Berm: Regrade to 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			WR Berm area included in Southwest Dump
A2.2.3.1		Revegetate	Seed/Fertilize: broadcast seeding	2.76	ha	C.6.01	91.4	$1,029.50	$2,839.68	$401.72	$1,108	$232.94	669	$642.53	$1,005.72	$2,774.08	$2,669.89	$7,364			
A2.2.3.2			Tree seedling application (1,000 stems/ha)	2.76	ha	C.6.06	141.4	$1,832.33	$5,054.12	$25.98	$72	$15.00	43	$41.37	$469.71	$1,295.60	$2,343.03	$6,463			
A2.3	*Reclamation Overburden Dump*																		$330,992		
A2.3.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	594	hrs	C.2.13	593.6	$44.39	$26,347.50	$194.68	$115,557	$67.01	41,432	$39,774.35	$0.00	$0.00	$306.07	$181,679			
A2.3.1.2			Slopes: Regrade to 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			Assumes borrow removed in such a manner that significant resloping is not required.
A2.3.2.1		Revegetate	Seed/Fertilize: broadcast seeding	29.79	ha	C.6.01	986.7	$1,029.50	$30,664.45	$401.72	$11,966	$232.94	7,228	$6,938.40	$1,005.72	$29,956.07	$2,669.89	$79,525			
A2.3.2.2			Tree seedling application (1,000 stems/ha)	29.79	ha	C.6.06	1,526.5	$1,832.33	$54,577.32	$25.98	$774	$15.00	465	$446.79	$469.71	$13,990.64	$2,343.03	$69,789			
A2.4	*Ridgetop South Backfill Dump*																		$34,189		
A2.4.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	17	hrs	C.2.13	16.9	$44.39	$751.05	$194.68	$3,294	$67.01	1,181	$1,133.78	$0.00	$0.00	$306.07	$5,179			
A2.4.1.2			Slopes: Regrade to 3H:1V or flatter	32	hrs	C.2.13	32.2	$44.39	$1,430.97	$194.68	$6,276	$67.01	2,250	$2,160.20	$0.00	$0.00	$306.07	$9,867			
A2.4.2.1		Revegetate	Seed/Fertilize: broadcast seeding	3.82	ha	C.6.01	126.5	$1,029.50	$3,931.30	$401.72	$1,534	$232.94	927	$889.53	$1,005.72	$3,840.48	$2,669.89	$10,195			
A2.4.2.2			Tree seedling application (1,000 stems/ha)	3.82	ha	C.6.06	195.7	$1,832.33	$6,997.02	$25.98	$99	$15.00	60	$57.28	$469.71	$1,793.65	$2,343.03	$8,947			
A3	**Ore Stockpiles**																			**$794,200**	
A3.1	*High and Medium Grade Sulphide Ore Stockpile (South of Mill)*																		$685,926		(North, West, East, and South Stockpiles)
A3.1.1.1		Relocate to Pit	Ore: Load, haul, dump, spread into Main Pit	0	m3	R.044	0.0	$0.52	$0.00	$1.72	$0	$0.57	0	$0.00	$0.00	$0.00	$2.82	$0			
A3.1.1.2			Ore Pads (over-excavate): Load, haul, dump spread into Main Pit	71,007	m3	R.044	835.3	$0.52	$36,800.28	$1.72	$122,356	$0.57	42,458	$40,759.51	$0.00	$0.00	$2.82	$199,915			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	EOM Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A3.1.1.3			Complete confirmation testing	1,421	ea	C.3.02	236.8	$7.43	$10,556.37	$1.33	$1,883	$0.00	0	$0.00	$13.26	$18,842.46	$22.01	$31,282			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	EOM Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A3.1.2.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	289	hrs	C.2.13	289.4	$44.39	$12,843.71	$194.68	$56,331	$67.01	20,197	$19,388.95	$0.00	$0.00	$306.07	$88,564			
A3.1.3.1		Cover	Cover material: Load, haul, dump spread (0.5m)	69,653	m3	R.055	1,264.9	$0.80	$55,711.18	$2.60	$181,043	$0.86	62,060	$59,577.97	$0.00	$0.00	$4.25	$296,332			
A3.1.4.1		Revegetate	Seed/Fertilize: broadcast seeding	13.93	ha	C.6.01	461.5	$1,029.50	$14,341.67	$401.72	$5,596	$232.94	3,380	$3,245.07	$1,005.72	$14,010.36	$2,669.89	$37,193			
A3.1.4.2			Tree seedling application (1,000 stems/ha)	13.93	ha	C.6.06	713.9	$1,832.33	$25,525.64	$25.98	$362	$15.00	218	$208.96	$469.71	$6,543.38	$2,343.03	$32,640			
A3.2	*Low Grade Sulphide Ore Stockpile (Main Dump Area)*																		$108,273		
A3.2.1.1		Relocate to Pit	Ore: Load, haul, dump, spread into Main Pit	0	m3	R.036	0.0	$0.56	$0.00	$1.84	$0	$0.62	0	$0.00	$0.00	$0.00	$3.02	$0			
A3.2.1.2			Ore Pad (over-excavate): Load, haul, dump spread into Main Pit	31,316	m3	R.036	396.1	$0.56	$17,435.36	$1.84	$57,732	$0.62	20,131	$19,325.41	$0.00	$0.00	$3.02	$94,493			
A3.2.1.3			Complete confirmation testing	626	ea	C.3.02	104.3	$7.43	$4,650.45	$1.33	$829	$0.00	0	$0.00	$13.26	$8,300.76	$22.01	$13,781			
A3.2.2.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			Included in Main Waste Dump Closure
A3.2.3.1		Cover	Cover material: Load, haul, dump spread	0	m3	R.050	0.0	$0.82	$0.00	$2.67	$0	$0.88	0	$0.00	$0.00	$0.00	$4.37	$0			Included in Main Waste Dump Closure
A3.2.4.1		Revegetate	Seed/Fertilize: broadcast seeding	0	ha	C.6.01	0.0	$1,029.50	$0.00	$401.72	$0	$232.94	0	$0.00	$1,005.72	$0.00	$2,669.89	$0			Included in Main Waste Dump Closure
A3.2.4.2			Tree seedling application (1,000 stems/ha)	0	ha	C.6.06	0.0	$1,832.33	$0.00	$25.98	$0	$15.00	0	$0.00	$469.71	$0.00	$2,343.03	$0			Included in Main Waste Dump Closure
A4	**Open Pits**																			**$620,582**	
A4.1	*Main Pit*																		$150,984		
A4.1.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	7,500	m2	C.2.03	28.5	$0.17	$1,265.55	$0.49	$3,658	$0.13	1,048	$1,005.87	$0.00	$0.00	$0.79	$5,930			
A4.1.1.2			Safety Berm: Construct around highwall perimeter	4,860	m3	C.2.02	45.8	$0.42	$2,035.14	$0.89	$4,333	$0.25	1,267	$1,215.92	$0.00	$0.00	$1.56	$7,584			
A4.1.1.3			Place large boulders across any pit access points	52	m3	R.001	1.6	$1.39	$72.20	$3.03	$157	$0.90	49	$46.76	$0.00	$0.00	$5.31	$276			
A4.1.1.4			Allowance: Install warning signs around pit perimeter at key locations	6	ea	C.5.13	5.6	$40.27	$241.63	$15.90	$95	$5.57	35	$33.39	$199.26	$1,195.57	$261.00	$1,566			
A4.1.2.1		Exposed tailings cover	Trafficability Layer: Load, haul, dump spread	17,750	m3	R.014	554.4	$1.38	$24,425.58	$2.86	$50,708	$0.82	15,091	$14,487.41	$0.00	$0.00	$5.05	$89,621			
A4.1.2.2			Overburden Layer: Load, haul, dump spread	8,875	m3	R.052	158.4	$0.79	$6,976.73	$2.55	$22,672	$0.84	7,772	$7,460.97	$0.00	$0.00	$4.18	$37,110			
A4.1.3.1		Revegetate	Seed/Fertilize: broadcast seeding	2	ha	C.6.01	58.8	$1,029.50	$1,827.37	$401.72	$713	$232.94	431	$413.48	$1,005.72	$1,785.15	$2,669.89	$4,739			
A4.1.3.2			Tree seedling application (1,000 stems/ha)	2	ha	C.6.06	91.0	$1,832.33	$3,252.39	$25.98	$46	$15.00	28	$26.63	$469.71	$833.74	$2,343.03	$4,159			
A4.2	*Area 2 Pit*																		$10,001		
A4.2.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	0	m2	C.2.03	0.0	$0.17	$0.00	$0.49	$0	$0.13	0	$0.00	$0.00	$0.00	$0.79	$0			
A4.2.1.2			Safety Berm: Construct around highwall perimeter	5,317	m3	C.2.02	50.2	$0.42	$2,226.54	$0.89	$4,740	$0.25	1,386	$1,330.27	$0.00	$0.00	$1.56	$8,297			
A4.2.1.3			Place large boulders across any pit access points	26	m3	R.001	0.8	$1.39	$36.10	$3.03	$79	$0.90	24	$23.38	$0.00	$0.00	$5.31	$138			
A4.2.1.4			Allowance: Install warning signs around pit perimeter at key locations	6	ea	C.5.13	5.6	$40.27	$241.63	$15.90	$95	$5.57	35	$33.39	$199.26	$1,195.57	$261.00	$1,566			
A4.3	*Area 118 Pit*																		$0		
A4.3.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	0	m2	C.2.03	0.0	$0.17	$0.00	$0.49	$0	$0.13	0	$0.00	$0.00	$0.00	$0.79	$0			Pit filled it by Area 118 Backfill Dump
A4.3.1.2			Safety Berm: Construct around highwall perimeter	0	m3	C.2.02	0.0	$0.42	$0.00	$0.89	$0	$0.25	0	$0.00	$0.00	$0.00	$1.56	$0			
A4.3.1.3			Place large boulders across any pit access points	0	m3	R.001	0.0	$1.39	$0.00	$3.03	$0	$0.90	0	$0.00	$0.00	$0.00	$5.31	$0			
A4.3.1.4			Allowance: Install warning signs around pit perimeter at key locations	0	ea	C.5.13	0.0	$40.27	$0.00	$15.90	$0	$5.57	0	$0.00	$199.26	$0.00	$261.00	$0			
A4.4	*Minto North Pit*																		$11,481		
A4.4.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	5,900	m2	C.2.03	22.4	$0.17	$995.56	$0.49	$2,878	$0.13	824	$791.29	$0.00	$0.00	$0.79	$4,665			
A4.4.1.2			Safety Berm: Construct around highwall perimeter	3,610	m	C.2.02	34.1	$0.42	$1,511.81	$0.89	$3,219	$0.25	941	$903.25	$0.00	$0.00	$1.56	$5,634			
A4.4.1.3			Place large boulders across any pit access points	26	m3	R.001	0.8	$1.39	$36.10	$3.03	$79	$0.90	24	$23.38	$0.00	$0.00	$5.31	$138			
A4.4.1.4			Allowance: Install warning signs around pit perimeter at key locations	4	ea	C.5.13	3.7	$40.27	$161.09	$15.90	$64	$5.57	23	$22.26	$199.26	$797.04	$261.00	$1,044			
A4.5	*Ridgetop North*																		$448,116		
A4.5.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	10,110	m2	C.2.03	38.4	$0.17	$1,705.95	$0.49	$4,931	$0.13	1,412	$1,355.92	$0.00	$0.00	$0.79	$7,993			
A4.5.1.2			Safety Berm: Construct around highwall perimeter	3,648	m	C.2.02	34.4	$0.42	$1,527.56	$0.89	$3,252	$0.25	951	$912.66	$0.00	$0.00	$1.56	$5,692			
A4.5.1.3			Place large boulders across any pit access points	26	m3	R.001	0.8	$1.39	$36.10	$3.03	$79	$0.90	24	$23.38	$0.00	$0.00	$5.31	$138			
A4.5.1.4			Allowance: Install warning signs around pit perimeter at key locations	6	ea	C.5.13	5.6	$40.27	$241.63	$15.90	$95	$5.57	35	$33.39	$199.26	$1,195.57	$261.00	$1,566			
A4.5.2.1		Exposed tailings cover	Trafficability Layer: Load, haul, dump spread	62,477	m3	R.068	816.5	$0.58	$35,941.63	$1.90	$119,010	$0.64	41,498	$39,837.81	$0.00	$0.00	$3.12	$194,789			
A4.5.2.2			Overburden Layer: Load, haul, dump spread	31,239	m3	R.062	882.0	$1.24	$38,844.69	$4.04	$126,233	$1.33	43,272	$41,540.81	$0.00	$0.00	$6.61	$206,618			
A4.5.3.1		Revegetate	Seed/Fertilize: broadcast seeding	6	ha	C.6.01	207.0	$1,029.50	$6,432.02	$401.72	$2,510	$232.94	1,516	$1,455.36	$1,005.72	$6,283.44	$2,669.89	$16,681			
A4.5.3.2			Tree seedling application (1,000 stems/ha)	6	ha	C.6.06	320.2	$1,832.33	$11,447.87	$25.98	$162	$15.00	98	$93.72	$469.71	$2,934.61	$2,343.03	$14,639			
A4.6	*Ridgetop South*																		$0		
A4.6.1.1		Secure Access	Safety Berm: Clear land around highwall perimeter (10m width)	0	m2	C.2.03	0.0	$0.17	$0.00	$0.49	$0	$0.13	0	$0.00	$0.00	$0.00	$0.79	$0			Pit to be backfilled
A4.6.1.2			Safety Berm: Construct around highwall perimeter	0	m	C.2.02	0.0	$0.42	$0.00	$0.89	$0	$0.25	0	$0.00	$0.00	$0.00	$1.56	$0			
A4.6.1.3			Place large boulders across any pit access points	0	m3	R.001	0.0	$1.39	$0.00	$3.03	$0	$0.90	0	$0.00	$0.00	$0.00	$5.31	$0			
A4.6.1.4			Allowance: Install warning signs around pit perimeter at key locations	0	ea	C.5.13	0.0	$40.27	$0.00	$15.90	$0	$5.57	0	$0.00	$199.26	$0.00	$261.00	$0			
A5	**Underground Openings**																			**$81,637**	
A5.1	*Minto North Portal*																		$16,573		
A5.1.1.1		Seal portal	Load, Haul, Dump backfill plug	2,588	m3	R.001	81.5	$1.39	$3,592.52	$3.03	$7,832	$0.90	2,423	$2,326.51	$0.00	$0.00	$5.31	$13,751			Material sourced from nearby laydown area rockfill
A5.1.1.2			Doze backfill plug into portal with small dozer	16	hrs	C.2.19	0.0	$44.39	$723.90	$105.36	$1,718	$23.28	395	$379.66	$0.00	$0.00	$173.02	$2,822			
A5.2	*Minto North Portal Egress*																		$16,573		
A5.2.1.1		Seal portal	Load, Haul, Dump backfill plug	2,588	m3	R.001	81.5	$1.39	$3,592.52	$3.03	$7,832	$0.90	2,423	$2,326.51	$0.00	$0.00	$5.31	$13,751			Material sourced from nearby laydown area rockfill
A5.2.1.2			Doze backfill plug into portal with small dozer	16	hrs	C.2.19	0.0	$44.39	$723.90	$105.36	$1,718	$23.28	395	$379.66	$0.00	$0.00	$173.02	$2,822			
A5.3	*Minto South Portal*																		$16,573		
A5.3.1.1		Seal portal	Load, Haul, Dump backfill plug	2,588	m3	R.001	81.5	$1.39	$3,592.52	$3.03	$7,832	$0.90	2,423	$2,326.51	$0.00	$0.00	$5.31	$13,751			Material sourced from nearby laydown area rockfill
A5.3.1.2			Doze backfill plug into portal with small dozer	16	hrs	C.2.19	0.0	$44.39	$723.90	$105.36	$1,718	$23.28	395	$379.66	$0.00	$0.00	$173.02	$2,822			
A5.3.2.1		Revegetate	Included in Yards (WBS No. A11-10)																		
A5.4	*Area 118 Vent Raise*																		$10,691		
A5.4.1.1		Site Preparation	Removal of Heater/Fan included in Demolition																		
A5.4.1.2			Excavate and clean perimeter around shaft to suitable material.	4	ea	C.8.01	8.0	$84.00	$335.99	$13.66	$55	$6.83	28	$27.33	$0.00	$0.00	$104.49	$418			
A5.4.2.1		Concrete Cap installation	Structural steel - Supply and install I-Beam structural support	8	m	C.8.02	9.8	$50.19	$401.48	$15.95	$128	$6.05	50	$48.37	$16.09	$128.68	$88.27	$706			
A5.4.2.2			Formwork: Supply and install concrete ring-wall formwork	8	m2	C.8.03	51.3	$309.43	$2,475.43	$0.00	$0	$0.00	0	$0.00	$3.54	$28.32	$312.97	$2,504			
A5.4.2.3			Ringwall Rebar: supply and install	32	m	C.8.05	0.8	$1.27	$40.74	$0.00	$0	$0.00	0	$0.00	$3.37	$107.82	$4.64	$149			
A5.4.2.4			Concrete slab: supply and install	8.6	m3	C.8.06	61.1	$329.23	$2,844.57	$10.63	$92	$3.50	31	$30.21	$386.22	$3,336.96	$729.58	$6,304			Includes slab formwork, rebar, finishing.
A5.4.2.5			Vent Raise Pipe Supply and install	1	ea.	C.8.07	0.5	$17.77	$17.77	$0.00	$0	$0.00	0	$0.00	$460.62	$460.62	$478.39	$478			
A5.4.3.1		Backfill cap area	Backfill: Load, haul, dump, place over concrete cap	25	m3	R.001	0.8	$1.39	$34.71	$3.03	$76	$0.90	23	$22.48	$0.00	$0.00	$5.31	$133			
A5.5	*Minto East Vent Raise*																		$10,610		
A5.5.1.2		Preparation	Excavate and clean perimeter around shaft to suitable material.	4	m3	C.8.01	8.0	$84.00	$335.99	$13.66	$55	$6.83	28	$27.33	$0.00	$0.00	$104.49	$418			
A5.5.2.1		Concrete Cap installation	Structural steel - Supply and install I-Beam structural support	8	m	C.8.02	9.8	$50.19	$401.48	$15.95	$128	$6.05	50	$48.37	$16.09	$128.68	$88.27	$706			
A5.5.2.2			Formwork: Supply and install concrete ring-wall formwork	8	m2	C.8.03	51.3	$309.43	$2,475.43	$0.00	$0	$0.00	0	$0.00	$3.54	$28.32	$312.97	$2,504			
A5.5.2.3			Ringwall Rebar: supply and install	14	m	C.8.05	0.3	$1.27	$18.33	$0.00	$0	$0.00	0	$0.00	$3.37	$48.52	$4.64	$67			
A5.5.2.4			Concrete slab: supply and install	8.6	m3	C.8.06	61.1	$329.23	$2,844.57	$10.63	$92	$3.50	31	$30.21	$386.22	$3,336.96	$729.58	$6,304			Includes slab formwork, rebar, finishing.
A5.5.2.5			Vent Raise Pipe Supply and install	1	ea.	C.8.07	0.5	$17.77	$17.77	$0.00	$0	$0.00	0	$0.00	$460.62	$460.62	$478.39	$478			
A5.5.3.1		Backfill cap area	Backfill: Load, haul, dump, place over concrete cap	25	m3	R.001	0.8	$1.39	$34.71	$3.03	$76	$0.90	23	$22.48	$0.00	$0.00	$5.31	$133			
A5.6	*Minto East 2 Vent Raise*																		$10,617		
A5.6.1.1		Site Preparation	Removal of Heater/Fan included in Demolition																		
A5.6.1.2			Excavate and clean perimeter around shaft to suitable material.	4	m3	C.8.01	8.0	$84.00	$335.99	$13.66	$55	$6.83	28	$27.33	$0.00	$0.00	$104.49	$418			
A5.6.2.1		Concrete Cap installation	Structural steel - Supply and install I-Beam structural support	8	m	C.8.02	9.8	$50.19	$401.48	$15.95	$128	$6.05	50	$48.37	$16.09	$128.68	$88.27	$706			
A5.6.2.2			Formwork: Supply and install concrete ring-wall formwork	8	m2	C.8.03	51.3	$309.43	$2,475.43	$0.00	$0	$0.00	0	$0.00	$3.54	$28.32	$312.97	$2,504			
A5.6.2.3			Ringwall Rebar: supply and install	16	m	C.8.05	0.4	$1.27	$20.37	$0.00	$0	$0.00	0	$0.00	$3.37	$53.91	$4.64	$74			
A5.6.2.4			Concrete slab: supply and install	8.6	m3	C.8.06	61.1	$329.23	$2,844.57	$10.63	$92	$3.50	31	$30.21	$386.22	$3,336.96	$729.58	$6,304			Includes slab formwork, rebar, finishing.
A5.6.2.5			Vent Raise Pipe Supply and install	1	ea.	C.8.07	0.5	$17.77	$17.77	$0.00	$0	$0.00	0	$0.00	$460.62	$460.62	$478.39	$478			
A5.6.3.1		Backfill cap area	Backfill: Load, haul, dump, place over concrete cap	25	m3	R.001	0.8	$1.39	$34.71	$3.03	$76	$0.90	23	$22.48	$0.00	$0.00	$5.31	$133			
A6	**External Tailings Facilities**																			**$57,291**	
A6.1	*Dry Stack Tailings Storage Facility*																		$57,291		
A6.1.1.1		Waste Rock Shell	Regrade WR shell slopes to 4H:1V	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	EOM Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A6.1.1.2			Cover material: Load, haul, dump spread (0.5m)	0	m3	R.048	0.0	$0.86	$0.00	$2.78	$0	$0.91	0	$0.00	$0.00	$0.00	$4.55	$0			
A6.1.1.3			Scarify surface	0	m2	C.2.16	0.0	$0.03	$0.00	$0.07	$0	$0.02	0	$0.00	$0.00	$0.00	$0.12	$0			
A6.1.1.4			Seed/Fertilize: broadcast seeding	0.00	ha	C.6.01	0.0	$1,029.50	$0.00	$401.72	$0	$232.94	0	$0.00	$1,005.72	$0.00	$2,669.89	$0			
A6.1.1.5			Tree seedling application (1,000 stems/ha)	5.30	ha	C.6.06	271.4	$1,832.33	$9,703.52	$25.98	$138	$15.00	83	$79.44	$469.71	$2,487.45	$2,343.03	$12,408			
A6.1.2.1		Fill Area at south end of DS	Load, haul, dump, spread	0	Cm3	R.001	0.0	$1.39	$0.00	$3.03	$0	$0.90	0	$0.00	$0.00	$0.00	$5.31	$0			Fill from TDD road and/or unsuitable DSTSF cover material
A6.1.3.1		Tailings surface	Regrade tailings area south of the tailings	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A6.1.3.2			Regrade surface to form tertiary drainage catchments	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A6.1.3.3			Cover: Excavate unsuitable temporary cover material	0	m3	R.003	0.0	$0.51	$0.00	$1.70	$0	$0.57	0	$0.00	$0.00	$0.00	$2.78	$0			
A6.1.3.4			Cover material: Load, haul, dump spread (0.5m)	0	m3	R.048	0.0	$0.86	$0.00	$2.78	$0	$0.91	0	$0.00	$0.00	$0.00	$4.55	$0			
A6.1.3.5			Seed/Fertilize: broadcast seeding	0.00	ha	C.6.01	0.0	$1,029.50	$0.00	$401.72	$0	$232.94	0	$0.00	$1,005.72	$0.00	$2,669.89	$0			
A6.1.3.6			Tree seedling application (1,000 stems/ha)	19.16	ha	C.6.06	981.8	$1,832.33	$35,100.35	$25.98	$498	$15.00	299	$287.34	$469.71	$8,997.81	$2,343.03	$44,883			
A7	**Roads**																			**$136,050**	
A7.1	*Exploration Roads*																		$33,589		
A7.1.1.1		Regrade/Scarify	Scarify road surface	54,285	m2	C.2.16	0.0	$0.03	$1,886.30	$0.07	$3,564	$0.02	965	$926.07	$0.00	$0.00	$0.12	$6,377			
A7.1.2.1		Revegetate	Seed/Fertilize: broadcast seeding	5.43	ha	C.6.01	179.8	$1,029.50	$5,588.65	$401.72	$2,181	$232.94	1,317	$1,264.54	$1,005.72	$5,459.55	$2,669.89	$14,494			
A7.1.2.2			Tree seedling application (1,000 stems/ha)	5.43	ha	C.6.06	278.2	$1,832.33	$9,946.82	$25.98	$141	$15.00	85	$81.43	$469.71	$2,549.82	$2,343.03	$12,719			
A7.2	*Access Roads (excl. Main Site Access Road)*																		$9,082		
A7.2.1.1		Regrade/Scarify	Regrade side slopes	9	hrs	C.2.13	9.5	$44.39	$419.85	$194.68	$1,841	$67.01	660	$633.81	$0.00	$0.00	$306.07	$2,895			
A7.2.1.2			Scarify road surface	52,636	m2	C.2.16	0.0	$0.03	$1,829.00	$0.07	$3,456	$0.02	935	$897.94	$0.00	$0.00	$0.12	$6,183			
A7.2.2.1		Revegetate	Seed/Fertilize: broadcast seeding	0.00	ha	C.6.01	0.0	$1,029.50	$0.77	$401.72	$0	$232.94	0	$0.17	$1,005.72	$0.75	$2,669.89	$2			
A7.2.2.2			Tree seedling application (1,000 stems/ha)	0.00	ha	C.6.06	0.0	$1,832.33	$1.36	$25.98	$0	$15.00	0	$0.01	$469.71	$0.35	$2,343.03	$2			
A7.3	*Main Site Access Road*																		$1,343		
A7.3.1.1		New Access over MVFE	Regrade new access through MVFE	0	m3																
A7.3.2.1		Signage	Install large sign on east and west side of barge landing on Yukon River	2	ea.	C.5.14	1.9	$40.27	$80.54	$15.90	$32	$5.57	12	$11.13	$609.74	$1,219.48	$671.48	$1,343			
A7.4	*Haul Roads*																		$72,436		
A7.4.1.1		Regrade/Scarify	Regrade side slopes	21	hrs	C.2.13	20.5	$44.39	$910.71	$194.68	$3,994	$67.01	1,432	$1,374.82	$0.00	$0.00	$306.07	$6,280			
A7.4.1.2			Scarify road surface	91,084	m2	C.2.16	0.0	$0.03	$3,164.99	$0.07	$5,981	$0.02	1,619	$1,553.84	$0.00	$0.00	$0.12	$10,700			
A7.4.2.1		Revegetate	Seed/Fertilize: broadcast seeding	11.06	ha	C.6.01	366.5	$1,029.50	$11,389.06	$401.72	$4,444	$232.94	2,684	$2,576.99	$1,005.72	$11,125.97	$2,669.89	$29,536			
A7.4.2.2			Tree seedling application (1,000 stems/ha)	11.06	ha	C.6.06	567.0	$1,832.33	$20,270.53	$25.98	$287	$15.00	173	$165.94	$469.71	$5,196.25	$2,343.03	$25,920			
A7.5	*Culvert Removal*																		$19,600		
A7.5.1.1		Excavate culverts	Mobilization allowance for excavation fleet to various sites	6	hrs	C.7.01	12.0	$88.22	$529.34	$118.29	$710	$44.16	276	$264.96	$0.00	$0.00	$250.67	$1,504			
A7.5.1.2			Load, haul, dump, spread (spoil locally)	18,020	m3	R.901	59.1	$0.15	$2,622.79	$0.58	$10,499	$0.20	3,841	$3,687.10	$0.00	$0.00	$0.93	$16,809			
A7.5.1.2		Waste disposal	On-site disposal (demolition debris, etc.)	190	Lm3	R.037	8.1	$1.87	$356.36	$3.75	$713	$1.14	227	$217.54	$0.00	$0.00	$6.77	$1,287			
A8	**Demolition**																			**$1,976,253**	
A8.1	*Airstrip Area*																		$21,059		
A8.1.1.1		Remove tanks/equipment	Dismantle and prep for transport	7	hrs	C.1.13	28.0	$168.00	$1,175.97	$86.19	$603	$14.53	106	$101.70	$0.00	$0.00	$268.71	$1,881			
A8.1.2.1		Waste Oil Tanker Secondar	Bedding Material: Load, haul dump bedding material to landfarm	47	Cm3	R.001	1.5	$1.39	$65.73	$3.03	$143	$0.90	44	$42.57	$0.00	$0.00	$5.31	$252			
A8.1.2.2			Cut and fold liner	219	m2	C.1.15	2.5	$0.43	$94.83	$0.19	$43	$0.07	16	$15.13	$0.00	$0.00	$0.70	$152			
A8.1.2.3			Regrade area to promote positive drainage	192	m2	C.2.11	1.9	$0.44	$85.22	$1.28	$246	$0.35	71	$67.74	$0.00	$0.00	$2.08	$399			
A8.1.3.1		Prepare for demolition	Remove hazardous materials/prep for transport offsite	10	hrs	C.1.22	40.0	$159.30	$1,592.98	$95.83	$958	$26.93	281	$269.28	$0.00	$0.00	$282.05	$2,821			
A8.1.3.2			Disconnect services	4	hrs	C.1.09	12.0	$88.07	$352.28	$7.55	$30	$4.80	20	$19.20	$0.00	$0.00	$100.42	$402			
A8.1.4.1		Demolition	Structural building demolition: wooden buildings/tents	1,149	m3	C.1.08	87.7	$2.88	$3,310.74	$1.60	$1,834	$0.57	680	$652.51	$0.00	$0.00	$5.05	$5,797			
A8.1.4.2			Other demolition: covered storage, debris, etc.	530	m3	C.1.07	52.5	$3.74	$1,980.53	$2.07	$1,097	$0.74	407	$390.34	$0.00	$0.00	$6.55	$3,468			
A8.1.5.1		Waste disposal	On-site disposal (demolition debris, etc.)	379	m3	R.006	12.8	$1.48	$562.06	$2.95	$1,117	$0.92	361	$346.89	$0.00	$0.00	$5.35	$2,026			
A8.1.5.2			Off-site disposal (re-usable equipment, etc.)	4	m3	C.7.07	34.6	$355.10	$1,536.93	$423.68	$1,834	$113.33	511	$490.50	$0.00	$0.00	$892.11	$3,861			
A8.2	*Airport Laydown Area*																		$33,408		
A8.2.1.1		Remove tanks/equipment	Dismantle and prep for transport	2	hrs	C.1.13	8.0	$168.00	$335.99	$86.19	$172	$14.53	30	$29.06	$0.00	$0.00	$268.71	$537			
A8.2.2.1		Prepare for demolition	Remove hazardous materials/prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.2.2.2			Disconnect services	4	hrs	C.1.09	12.0	$88.07	$352.28	$7.55	$30	$4.80	20	$19.20	$0.00	$0.00	$100.42	$402			
A8.2.3.1		Demolition	Structural building demolition: steel structures	96	tonnes	C.1.05	191.3	$69.36	$6,633.00	$139.08	$13,301	$33.84	3,371	$3,236.25	$0.00	$0.00	$242.28	$23,170			
A8.2.3.2			Other demolition: covered storage etc.	560	m3	C.1.07	55.5	$3.74	$2,093.92	$2.07	$1,160	$0.74	430	$412.69	$0.00	$0.00	$6.55	$3,667			
A8.2.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	393	Lm3	R.005	15.8	$1.76	$693.24	$3.53	$1,387	$1.08	441	$423.19	$0.00	$0.00	$6.37	$2,504			
A8.2.4.2			Off-site disposal (re-usable equipment, etc.)	3	trips	C.7.07	23.0	$355.10	$1,020.92	$423.68	$1,218	$113.33	339	$325.82	$0.00	$0.00	$892.11	$2,565			
A8.3	*Camp Area*																		$375,774		
A8.3.1.1		Equipment Removal	Remove salvageable equipment	48	hrs	C.1.13	192.0	$168.00	$8,063.76	$86.19	$4,137	$14.53	726	$697.34	$0.00	$0.00	$268.71	$12,898			
A8.3.2.1		Remove modular buildings	Dismantle and prep for transport	92	ea.	C.1.20	1,794.0	$850.98	$78,290.06	$730.60	$67,215	$166.46	15,953	$15,314.69	$0.00	$0.00	$1,748.04	$160,820			
A8.3.2.2			Transport structures off-site (Whitehorse)	92	ea.	C.7.04	736.0	$351.24	$32,314.23	$303.64	$27,935	$113.33	10,861	$10,426.18	$0.00	$0.00	$768.21	$70,675			
A8.3.3.1		Prepare for demolition	Remove hazardous materials/prep for transport offsite	1	hrs	C.1.22	4.0	$159.30	$159.30	$95.83	$96	$26.93	28	$26.93	$0.00	$0.00	$282.05	$282			
A8.3.3.2			Disconnect services	8	hrs	C.1.09	24.0	$88.07	$704.56	$7.55	$60	$4.80	40	$38.40	$0.00	$0.00	$100.42	$803			
A8.3.4.1		Demolition	Structural building demolition: Steel	175	tonnes	C.1.05	350.7	$69.36	$12,160.50	$139.08	$24,385	$33.84	6,180	$5,933.13	$0.00	$0.00	$242.28	$42,479			
A8.3.4.2			Structural building demolition: Wood/misc. structures	10,959	m3	C.1.08	836.7	$2.88	$31,584.20	$1.60	$17,497	$0.57	6,484	$6,224.86	$0.00	$0.00	$5.05	$55,306			
A8.3.4.4			Other demolition: Utilidors, etc.	466	m3	C.1.07	46.1	$3.74	$1,742.09	$2.07	$965	$0.74	358	$343.34	$0.00	$0.00	$6.55	$3,051			
A8.3.5.1		Waste disposal	On-site disposal (demolition debris, etc.)	3,867	m3	R.007	175.3	$1.99	$7,708.84	$3.99	$15,425	$1.22	4,902	$4,705.86	$0.00	$0.00	$7.20	$27,839			
A8.3.5.2			Off-site disposal (re-usable equipment, etc.)	1	trips	C.7.07	8.0	$355.10	$355.10	$423.68	$424	$113.33	118	$113.33	$0.00	$0.00	$892.11	$892			
A8.3.6.1		Demolish foundations	Break in place concrete foundations	51	m3	C.1.01	4.1	$2.39	$122.15	$9.37	$479	$2.48	132	$126.81	$0.00	$0.00	$14.24	$728			
A8.4	*Explosives Plant and Storage Areas*																		$30,611		
A8.4.1.1		Remove equipment	Small equipment: dismantle and prep for transport	37	hrs	C.1.13	149.3	$168.00	$6,271.81	$86.19	$3,218	$14.53	565	$542.38	$0.00	$0.00	$268.71	$10,032			
A8.4.2.1		Remove modular buildings	Dismantle and prep for transport	1	ea.	C.1.20	19.5	$850.98	$850.98	$730.60	$731	$166.46	173	$166.46	$0.00	$0.00	$1,748.04	$1,748			
A8.4.2.2			Transport structures off-site (Whitehorse)	1	ea.	C.7.04	8.0	$351.24	$351.24	$303.64	$304	$113.33	118	$113.33	$0.00	$0.00	$768.21	$768			
A8.4.3.1		Prepare for demolition	Remove hazardous materials/prep for transport offsite	4	hrs	C.1.22	16.0	$159.30	$637.19	$95.83	$383	$26.93	112	$107.71	$0.00	$0.00	$282.05	$1,128			
A8.4.3.2			Disconnect services	2	hrs	C.1.09	6.0	$88.07	$176.14	$7.55	$15	$4.80	10	$9.60	$0.00	$0.00	$100.42	$201			
A8.4.4.1		Demolition	Structural building demolition: Steel	47	tonnes	C.1.05	94.2	$69.36	$3,266.25	$139.08	$6,550	$33.84	1,660	$1,593.61	$0.00	$0.00	$242.28	$11,410			
A8.4.5.1		Waste disposal	On-site disposal (demolition debris, etc.)	103	m3	R.008	5.4	$2.32	$239.23	$4.65	$479	$1.42	152	$146.04	$0.00	$0.00	$8.39	$864			
A8.4.5.2			Off-site disposal (re-usable equipment, etc.)	5	trips	C.7.07	40.0	$355.10	$1,775.51	$423.68	$2,118	$113.33	590	$566.64	$0.00	$0.00	$892.11	$4,461			
A8.5	*Fuel Storage Area*																		$114,842		
A8.5.1.1		Remove equipment/dismant	Small equipment: dismantle and prep for transport	24	hrs	C.1.13	96.0	$168.00	$4,031.88	$86.19	$2,069	$14.53	363	$348.67	$0.00	$0.00	$268.71	$6,449			
A8.5.1.2			Large equipment (crane req'd): dismantle and prep for transport	192	hrs	C.1.14	1,153.4	$269.89	$51,883.26	$167.56	$32,213	$20.53	4,111	$3,946.31	$0.00	$0.00	$457.98	$88,042			
A8.5.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.5.2.2			Disconnect services	8	hrs	C.1.09	24.0	$88.07	$704.56	$7.55	$60	$4.80	40	$38.40	$0.00	$0.00	$100.42	$803			
A8.5.2.3			Clean out tanks, pressure wash, remove sludge	10	hrs	C.1.04	10.3	$35.54	$365.13	$15.75	$162	$2.00	21	$20.55	$0.00	$0.00	$53.29	$548			
A8.5.3.1		Demolish structures	Misc. debris/scraps (tanks included in dismantling above)	75	m3	C.1.08	5.7	$2.88	$216.15	$1.60	$120	$0.57	44	$42.60	$0.00	$0.00	$5.05	$378			
A8.5.4.1		Remove secondary containi	Bedding Material: Load, haul dump bedding material to landfarm	623	Cm3	R.009	32.8	$2.32	$1,442.73	$4.64	$2,887	$1.41	917	$880.71	$0.00	$0.00	$8.37	$5,210			
A8.5.4.2			Cut and fold liner	1,747	m2	C.1.15	20.3	$0.43	$758.23	$0.19	$340	$0.07	126	$120.97	$0.00	$0.00	$0.70	$1,219			
A8.5.4.3			Regrade area to promote positive drainage	1,666	m2	C.2.11	16.7	$0.44	$739.50	$1.28	$2,138	$0.35	612	$587.76	$0.00	$0.00	$2.08	$3,465			
A8.5.5.1		Waste disposal	On-site disposal (demolition debris, etc.)	229	m3	R.009	12.1	$2.32	$531.18	$4.64	$1,063	$1.41	338	$324.26	$0.00	$0.00	$8.37	$1,918			
A8.5.5.2			Off-site disposal (re-usable equipment, etc.)	7	trips	C.7.07	56.0	$355.10	$2,485.71	$423.68	$2,966	$113.33	826	$793.30	$0.00	$0.00	$892.11	$6,245			
A8.6	*Mill Area*																		$1,265,173		Excludes WTP
A8.6.1.1		Remove equipment	Small equipment: dismantle and prep for transport	379	hrs	C.1.13	1,515.0	$168.00	$63,626.87	$86.19	$32,644	$14.53	5,732	$5,502.37	$0.00	$0.00	$268.71	$101,773			
A8.6.1.2			Large equipment (crane req'd): dismantle and prep for transport	466	hrs	C.1.14	2,796.5	$269.89	$125,789.65	$167.56	$78,099	$20.53	9,966	$9,567.74	$0.00	$0.00	$457.98	$213,456			
A8.6.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	65	hrs	C.1.22	260.0	$159.30	$10,354.37	$95.83	$6,229	$26.93	1,823	$1,750.32	$0.00	$0.00	$282.05	$18,333			Off-site disposal included in 'Waste Disposal'
A8.6.2.6			Reagents: Disposal and tipping fees	1.0	ls	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	EOM Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A8.6.2.7			Decontaminate buildings: wash equipment/structures, etc.	38	hrs	C.1.04	37.8	$35.54	$1,342.93	$15.75	$595	$2.00	79	$75.58	$0.00	$0.00	$53.29	$2,014			
A8.6.2.8			Decontaminate buildings: Wash floors etc.	4,732	m2	C.1.03	40.8	$0.31	$1,448.16	$0.14	$642	$0.02	85	$81.50	$0.00	$0.00	$0.46	$2,171			
A8.6.3.1		Demolition	Structural building demolition: Steel	2,660	tonnes	C.1.05	5,319.1	$69.36	$184,463.18	$139.08	$369,899	$33.84	93,750	$89,999.96	$0.00	$0.00	$242.28	$644,363			
A8.6.3.2			Structural building demolition: Wood/misc. structures	960	m3	C.1.08	73.3	$2.88	$2,766.75	$1.60	$1,533	$0.57	568	$545.29	$0.00	$0.00	$5.05	$4,845			
A8.6.3.3			Other demolition: Steel	526	tonnes	C.1.05	1,052.7	$69.36	$36,505.48	$139.08	$73,204	$33.84	18,553	$17,811.10	$0.00	$0.00	$242.28	$127,520			
A8.6.3.4			Other demolition: concrete	602	m3	C.1.06	323.7	$15.92	$9,579.30	$62.48	$37,592	$16.53	10,359	$9,944.51	$0.00	$0.00	$94.92	$57,116			
A8.6.3.5			Other demolition: miscellaneous	244	m3	C.1.07	24.1	$3.74	$910.78	$2.07	$505	$0.74	187	$179.50	$0.00	$0.00	$6.55	$1,595			
A8.6.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	5,985	m3	R.037	255.1	$1.87	$11,217.23	$3.75	$22,445	$1.14	7,133	$6,847.56	$0.00	$0.00	$6.77	$40,510			
A8.6.4.2			Off-site disposal (re-usable equipment, etc.)	33	trips	C.7.07	266.0	$355.10	$11,806.23	$423.68	$14,086	$113.33	3,925	$3,767.87	$0.00	$0.00	$892	$29,660			
A8.6.5.1		Demolish foundations	Break in place concrete foundations	1,532	m3	C.1.01	123.7	$2.39	$3,659.05	$9.37	$14,359	$2.48	3,957	$3,798.55	$0.00	$0.00	$14.24	$21,817			
A8.7	*Mill Valley Fill Extension (Stage 1 and 2)*																		$28,525		
A8.7.1.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	8	hrs	C.1.22	30.0	$159.30	$1,194.73	$95.83	$719	$26.93	210	$201.96	$0.00	$0.00	$282.05	$2,115			
A8.7.1.2			Reagents: Disposal and tipping fees	1	ls	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A8.7.2.1		Demolish buildings	Structural building demolition: Wood/misc. structures (Reagent tent)	1,800	m3	C.1.08	137.4	$2.88	$5,187.66	$1.60	$2,874	$0.57	1,065	$1,022.42	$0.00	$0.00	$5.05	$9,084			
A8.7.3.1		Haul waste to landfill	Demolition debris	594	m3	R.037	25.3	$1.87	$1,113.25	$3.75	$2,228	$1.14	708	$679.58	$0.00	$0.00	$6.77	$4,020			
A8.7.3.2			Remove tires off-site for disposal	10	trips	C.7.07	78.8	$355.10	$3,495.53	$423.68	$4,171	$113.33	1,162	$1,115.57	$0.00	$0.00	$892.11	$8,782			
A8.7.3.3			Misc. debris/scraps	652	m3	R.037	27.8	$1.87	$1,221.95	$3.75	$2,445	$1.14	777	$745.94	$0.00	$0.00	$6.77	$4,413			
A8.7.4.1		Demolish foundations	Break and bury concrete foundations	8	m3	C.1.01	0.6	$2.39	$18.48	$9.37	$73	$2.48	20	$19.19	$0.00	$0.00	$14.24	$110			
A8.8	*Minto South Portal*																		$8,516		
A8.8.1.1		Remove equipment	Small equipment: dismantle and prep for transport	6	hrs	C.1.13	24.0	$168.00	$1,007.97	$86.19	$517	$14.53	91	$87.17	$0.00	$0.00	$268.71	$1,612			
A8.8.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.8.3.1		Demolition	Other demolition: miscellaneous	462	m3	C.1.07	45.7	$3.74	$1,727.03	$2.07	$957	$0.74	355	$340.38	$0.00	$0.00	$6.55	$3,024			
A8.8.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	462	m3	R.043	20.9	$1.99	$918.18	$3.98	$1,837	$1.21	584	$560.51	$0.00	$0.00	$7.18	$3,316			
A8.9	*Minto North Portal*																		$8,981		
A8.9.1.1		Remove equipment	Small equipment: dismantle and prep for transport	6	hrs	C.1.13	24.0	$168.00	$1,007.97	$86.19	$517	$14.53	91	$87.17	$0.00	$0.00	$268.71	$1,612			Qnty assumed to be same as Minto South Portal
A8.9.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			Qnty assumed to be same as Minto South Portal
A8.9.3.1		Demolition	Other demolition: miscellaneous	462	m3	C.1.07	45.7	$3.74	$1,727.03	$2.07	$957	$0.74	355	$340.38	$0.00	$0.00	$6.55	$3,024			Qnty assumed to be same as Minto South Portal
A8.9.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	462	m3	R.042	23.8	$2.27	$1,046.71	$4.53	$2,094	$1.38	666	$638.96	$0.00	$0.00	$8.18	$3,780			Qnty assumed to be same as Minto South Portal
A8.10	*Pelly Laydown (Includes propane tanks by W15 Sump)*																		$83,981		
A8.10.1.1		Remove modular buildings/t	Dismantle and prep for transport	7	ea.	C.1.20	136.5	$850.98	$5,956.85	$730.60	$5,114	$166.46	1,214	$1,165.25	$0.00	$0.00	$1,748.04	$12,236			
A8.10.1.2			Transport structures off-site (Whitehorse)	7	ea.	C.7.04	56.0	$351.24	$2,458.69	$303.64	$2,125	$113.33	826	$793.30	$0.00	$0.00	$768.21	$5,377			
A8.10.2.1		Prepare for demolition	Hazardous materials: Gather and prep for transport offsite	2	hrs	C.1.22	8.0	$159.30	$318.60	$95.83	$192	$26.93	56	$53.86	$0.00	$0.00	$282.05	$564			
A8.10.3.1		Demolish buildings	Structural building demolition: Steel	80	tonnes	C.1.05	159.4	$69.36	$5,527.50	$139.08	$11,084	$33.84	2,809	$2,696.88	$0.00	$0.00	$242.28	$19,309			
A8.10.3.2			Structural building demolition: Wood/misc. structures	5,672	m3	C.1.08	433.1	$2.88	$16,347.99	$1.60	$9,057	$0.57	3,356	$3,221.99	$0.00	$0.00	$5.05	$28,627			
A8.10.4.1		Waste disposal	On-site disposal (demolition debris, etc.)	2,165	m3	R.045	100.0	$2.03	$4,395.81	$4.06	$8,796	$1.24	2,795	$2,683.42	$0.00	$0.00	$7.33	$15,875			
A8.10.4.2			Off-site disposal (re-usable equipment, etc.)	2	trips	C.7.07	15.0	$355.10	$665.82	$423.68	$794	$113.33	221	$212.49	$0.00	$0.00	$892.11	$1,673			
A8.10.5.1		Demolish foundations	Break and bury concrete foundations	23	m3	C.1.01	1.8	$2.39	$53.73	$9.37	$211	$2.48	58	$55.78	$0.00	$0.00	$14.24	$320			
A8.11	*Vent Raise - Area 118*																		$5,381		
A8.11.1.1		Remove equipment	Small equipment: dismantle and prep for transport	20	hrs	C.1.13	80.0	$168.00	$3,359.90	$86.19	$1,724	$14.53	303	$290.56	$0.00	$0.00	$268.71	$5,374			
A8.11.2.1		Waste disposal	Off-site disposal (re-usable equipment, etc.)	1	trips	R.043	0.0	$1.99	$1.99	$3.98	$4	$1.21	1	$1.21	$0.00	$0.00	$7.18	$7			
A9	**Surface Infrastructure**																			**$169,527**	
A9.1	*Tailings and Water Conveyance Pipelines*																		$91,886		
A9.1.1.1		Dismantle piping systems	Flush and clean tailings pipeline systems	8	ls	C.1.21	32.0	$203.78	$1,630.27	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$203.78	$1,630			
A9.1.1.2			Cut pipelines and prep for transport (6 to 8" HDPE pipes)	6,800	m	C.1.17	1,070.9	$7.21	$49,056.06	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$7.21	$49,056			
A9.1.1.3			Cut pipelines and prep for transport (10 to 18" HDPE pipes)	1,740	m	C.1.18	456.7	$12.02	$20,920.96	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$12.02	$20,921			
A9.1.1.4			Cut pipelines and prep for transport (20 to 36" HDPE pipes)	885	m	C.1.19	348.4	$18.04	$15,961.25	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$18.04	$15,961			
A9.2.2.1		Waste disposal	Haul pipes lines and debris to landfill	638	m3	R.037	27.2	$1.87	$1,195.62	$3.75	$2,392	$1.14	760	$729.86	$0.00	$0.00	$6.77	$4,318			
A9.2	*Conveyance Pipeline Equipment*																		$12,462		
A9.2.1.1		Dismantle piping systems	Remove pumps and prep for transport	15	ea.	C.1.24	198.0	$602.98	$9,044.72	$24.94	$374	$14.40	225	$216.00	$0.00	$0.00	$642.33	$9,635			
A9.2.1.2			Remove barges/Support Equipment prep for transport	2	ea.	C.1.23	52.8	$1,205.96	$2,411.93	$49.89	$100	$28.80	60	$57.60	$0.00	$0.00	$1,284.65	$2,569			
A9.2.2.1		Waste disposal	Transport pumps, barge and support equip. offsite (Whitehorse)	6.7	tonnes	C.7.02	2.7	$17.56	$117.67	$15.18	$102	$5.67	40	$37.96	$0.00	$0.00	$38.41	$257			
A9.3	*Powerlines*																		$65,179		
A9.3.1.1		Dismantle	Dismantle and collect powerlines	3.2	km	C.1.10	217.9	$2,639.96	$8,342.27	$940.09	$2,971	$251.90	829	$795.99	$0.00	$0.00	$3,831.94	$12,109			
A9.3.1.2			Remove power poles and load for transport off-site	60	ea.	C.1.11	548.6	$423.67	$25,420.14	$88.51	$5,311	$18.29	1,143	$1,097.14	$0.00	$0.00	$530.46	$31,828			
A9.3.1.3			Disconnect transformers and load for transport	4	ea.	C.1.12	93.3	$892.89	$3,571.55	$317.96	$1,272	$85.20	355	$340.78	$0.00	$0.00	$1,296.04	$5,184			
A9.3.2.1		Waste disposal	Transport powerlines, poles and transformers off-site	18	trips	C.7.07	144.0	$355.10	$6,391.83	$423.68	$7,626	$113.33	2,125	$2,039.90	$0.00	$0.00	$892.11	$16,058			
A10	**Water Detention Structures**																			**$786,952**	
A10.1	*W15 Sump*																		$1,049		
A10.1.1.1		Remove secondary containr	Cut and fold liner	1,310	m2	C.1.15	15.2	$0.43	$568.50	$0.19	$255	$0.07	94	$90.70	$0.00	$0.00	$0.70	$914			
A10.1.1.2			Haul liner to landfill	20	Lm3	R.070	0.8	$1.90	$37.35	$3.80	$75	$1.16	24	$22.80	$0.00	$0.00	$6.86	$135			
A10.2	*Mill Water Pond*																		$0		
A10.3	*Sewage Lagoon (near IROD)*																		$23,852		
A10.3.1		Remove pond	Backfill pond	4,488	m3	R.001	141.4	$1.39	$6,231.19	$3.03	$13,585	$0.90	4,203	$4,035.32	$0.00	$0.00	$5.31	$23,852			
A10.4	*Water Storage Pond Dam*																		$762,051		
A10.4.1.1		Water Management	Install pump and pump around system	4	hrs	C.4.14	16.0	$120.43	$481.73	$20.53	$82	$6.80	28	$27.20	$0.00	$0.00	$147.76	$591			
A10.4.1.2			Pump pond water to discharge	400,000	m3	C.4.12	146.8	$0.01	$5,216.05	$0.05	$19,052	$0.01	3,058	$2,935.56	$0.00	$0.00	$0.07	$27,203			
A10.4.1.3			Maintain pump around system during WSP closure activities	103	days	C.4.13	10.3	$3.55	$364.31	$12.98	$1,331	$2.00	214	$205.03	$0.00	$0.00	$18.53	$1,900			Duration calculated as sum of associated tasks plus 20% contingency
A10.4.2.1		Breach dam	Granular zones/gen. fill: Load, haul, dump, spread in upstream North ab	62,301	m3	R.001	1,963.4	$1.39	$86,500.02	$3.03	$188,587	$0.90	58,351	$56,017.38	$0.00	$0.00	$5.31	$331,104			
A10.4.2.2			Dam Core: Load, haul, dump in temporary stockpile	23,617	m3	R.001	744.3	$1.39	$32,789.73	$3.03	$71,488	$0.90	22,119	$21,234.62	$0.00	$0.00	$5.31	$125,512			
A10.4.2.3			Rip-rap: Sort and stockpile durable rip-rap for reuse	19,285	m3	R.001	607.8	$1.39	$26,776.14	$3.03	$58,377	$0.90	18,063	$17,340.22	$0.00	$0.00	$5.31	$102,494			
A10.4.2.4			Rip-rap: Load haul, dump unsuitable rip-rap in upstream North abutmen	19,285	m3	R.001	607.8	$1.39	$26,776.14	$3.03	$58,377	$0.90	18,063	$17,340.22	$0.00	$0.00	$5.31	$102,494			
A10.4.3.1		Channel restoration	Excavate stream channel	5,584	m3	R.902	97.1	$0.77	$4,308.58	$1.64	$9,173	$0.46	2,681	$2,574.22	$0.00	$0.00	$2.88	$16,056			
A10.4.3.2			Bedding layer: Load haul, dump, place granular bedding layer along cha	726	m3	R.001	22.9	$1.39	$1,008.60	$3.03	$2,199	$0.90	680	$653.17	$0.00	$0.00	$5.31	$3,861			
A10.4.3.3			Rip-rap: Load, haul dump from temporary stockpile	964	m3	R.001	30.4	$1.39	$1,337.77	$3.03	$2,917	$0.90	902	$866.34	$0.00	$0.00	$5.31	$5,121			
A10.4.3.4			Rip-rap: place and secure	964	m3	R.904	10.5	$0.48	$465.10	$0.91	$877	$0.32	325	$311.83	$0.00	$0.00	$1.72	$1,653			
A10.4.4.1		Cover	Cover material: Load, haul, dump spread (1m)	6,975	m3	R.001	219.8	$1.39	$9,684.17	$3.03	$21,113	$0.90	6,533	$6,271.47	$0.00	$0.00	$5.31	$37,069			Cover sourced from dam core material
A10.4.5.1		Revegetate	Seed/Fertilize: broadcast seeding	1.4	ha	C.6.01	46.2	$1,029.50	$1,436.16	$401.72	$560	$232.94	338	$324.96	$1,005.72	$1,402.98	$2,669.89	$3,724			
A10.4.5.2			Tree seedling application (1,000 stems/ha)	1.4	ha	C.6.06	71.5	$1,832.33	$2,556.10	$25.98	$36	$15.00	22	$20.93	$469.71	$655.25	$2,343.03	$3,269			
A11	**Yards/Laydown Areas**																			**$758,907**	
A11.1	*Airstrip Area*																		$39,834		
A11.1.1.1		Re-grade	Re-grade slopes to be 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A11.1.1.2			Scarify surface	64,378	m2	C.2.16	0.0	$0.03	$2,236.99	$0.07	$4,227	$0.02	1,144	$1,098.24	$0.00	$0.00	$0.12	$7,562			
A11.1.2.1		Revegetate	Seed/Fertilize: broadcast seeding	6.4	ha	C.6.01	213.3	$1,029.50	$6,627.68	$401.72	$2,586	$232.94	1,562	$1,499.64	$1,005.72	$6,474.57	$2,669.89	$17,188			
A11.1.2.2			Tree seedling application (1,000 stems/ha)	6.4	ha	C.6.06	329.9	$1,832.33	$11,796.10	$25.98	$167	$15.00	101	$96.57	$469.71	$3,023.88	$2,343.03	$15,084			
A11.2	*Airport Laydown Area*																		$13,165		
A11.2.1.1		Re-grade	Scarify surface	21,277	m2	C.2.16	0.0	$0.03	$739.33	$0.07	$1,397	$0.02	378	$362.97	$0.00	$0.00	$0.12	$2,499			
A11.2.2.1		Revegetate	Seed/Fertilize: broadcast seeding	2.1	ha	C.6.01	70.5	$1,029.50	$2,190.47	$401.72	$855	$232.94	516	$495.64	$1,005.72	$2,139.87	$2,669.89	$5,681			
A11.2.2.2			Tree seedling application (1,000 stems/ha)	2.1	ha	C.6.06	109.0	$1,832.33	$3,898.65	$25.98	$55	$15.00	33	$31.92	$469.71	$999.40	$2,343.03	$4,985			
A11.3	*Camp Area*																		$118,708		
A11.3.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	7	hrs	C.2.13	6.9	$44.39	$306.20	$194.68	$1,343	$67.01	481	$462.24	$0.00	$0.00	$306.07	$2,111			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	EOM Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A11.3.1.2			Re-grade slopes to be 3H:1V or flatter	24	hrs	C.2.13	24.5	$44.39	$1,086.58	$194.68	$4,766	$67.01	1,709	$1,640.31	$0.00	$0.00	$306.07	$7,493			Slope between camp and main access road
A11.3.1.3			Scarify surfaces	20,626	m2	C.2.16	0.0	$0.03	$716.72	$0.07	$1,354	$0.02	367	$351.87	$0.00	$0.00	$0.12	$2,423			
A11.3.2.1		Cover	Flat area cover material: Load, haul, dump spread	10,313	Cm3	R.046	198.2	$0.85	$8,729.48	$2.75	$28,368	$0.91	9,724	$9,335.37	$0.00	$0.00	$4.50	$46,433			
A11.3.2.2			Slope area cover material: Load, haul, dump along crest	9,349	Cm3	R.047	157.2	$0.74	$6,916.70	$2.44	$22,835	$0.82	7,990	$7,670.53	$0.00	$0.00	$4.00	$37,422			
A11.3.2.3			Slope area cover material: Spread down slope	10	hrs	C.2.13	10.2	$44.39	$451.46	$194.68	$1,980	$67.01	710	$681.52	$0.00	$0.00	$306.07	$3,113			
A11.3.3.1		Revegetate	Seed/Fertilize: broadcast seeding	3.93	ha	C.6.01	130.3	$1,029.50	$4,048.49	$401.72	$1,580	$232.94	954	$916.05	$1,005.72	$3,954.97	$2,669.89	$10,499			
A11.3.3.2			Tree seedling application (1,000 stems/ha)	3.93	ha	C.6.06	201.5	$1,832.33	$7,205.60	$25.98	$102	$15.00	61	$58.99	$469.71	$1,847.12	$2,343.03	$9,214			
A11.4	*Crusher Area*																		$73,635		
A11.2.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	9	hrs	C.2.13	9.0	$44.39	$400.28	$194.68	$1,756	$67.01	629	$604.27	$0.00	$0.00	$306.07	$2,760			
A11.2.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	13,482	m3	R.055	244.8	$0.80	$10,783.37	$2.60	$35,042	$0.86	12,012	$11,531.82	$0.00	$0.00	$4.25	$57,358			
A11.2.3.1		Revegetate	Seed/Fertilize: broadcast seeding	2.70	ha	C.6.01	89.3	$1,029.50	$2,775.95	$401.72	$1,083	$232.94	654	$628.11	$1,005.72	$2,711.82	$2,669.89	$7,199			
A11.2.3.2			Tree seedling application (1,000 stems/ha)	3	ha	C.6.06	138.2	$1,832.33	$4,940.70	$25.98	$70	$15.00	42	$40.45	$469.71	$1,266.53	$2,343.03	$6,318			
A11.5	*Exploration Disturbances*																		$21,519		
A11.5.1.1		Ridgetop Area	Scarify surfaces	34,778	m2	C.2.16	0.0	$0.03	$1,208.47	$0.07	$2,284	$0.02	618	$593.29	$0.00	$0.00	$0.12	$4,085			
A11.5.1.2			Seed/Fertilize: broadcast seeding	3	ha	C.6.01	115.2	$1,029.50	$3,580.40	$401.72	$1,397	$232.94	844	$810.13	$1,005.72	$3,497.69	$2,669.89	$9,285			
A11.5.1.3			Tree seedling application (1,000 stems/ha)	3	ha	C.6.06	178.2	$1,832.33	$6,372.49	$25.98	$90	$15.00	54	$52.17	$469.71	$1,633.56	$2,343.03	$8,149			
A11.6	*Explosives Plant and Storage Areas*																		$43,165		
A11.6.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	30	hrs	C.2.13	29.6	$44.39	$1,314.39	$194.68	$5,765	$67.01	2,067	$1,984.22	$0.00	$0.00	$306.07	$9,063			
A11.6.1.2			Re-grade slopes to be 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A11.6.1.3			Scarify surfaces	55,113	m2	C.2.16	0.0	$0.03	$1,915.08	$0.07	$3,619	$0.02	979	$940.20	$0.00	$0.00	$0.12	$6,474			
A11.6.2.1		Revegetate	Seed/Fertilize: broadcast seeding	5.5	ha	C.6.01	182.6	$1,029.50	$5,673.94	$401.72	$2,214	$232.94	1,337	$1,283.83	$1,005.72	$5,542.86	$2,669.89	$14,715			
A11.6.2.2			Tree seedling application (1,000 stems/ha)	5.5	ha	C.6.06	282.5	$1,832.33	$10,098.61	$25.98	$143	$15.00	86	$82.67	$469.71	$2,588.73	$2,343.03	$12,913			
A11.7	*Fuel Storage Areas*																		$24,758		
A11.7.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	5	hrs	C.2.13	4.5	$44.39	$200.33	$194.68	$879	$67.01	315	$302.42	$0.00	$0.00	$306.07	$1,381			
A11.7.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	4,200	m3	R.049	81.8	$0.86	$3,603.19	$2.79	$11,709	$0.92	4,014	$3,853.28	$0.00	$0.00	$4.56	$19,166			
A11.7.3.1		Revegetate	Seed/Fertilize: broadcast seeding	0.84	ha	C.6.01	27.8	$1,029.50	$864.78	$401.72	$337	$232.94	204	$195.67	$1,005.72	$844.80	$2,669.89	$2,243			
A11.7.3.2			Tree seedling application (1,000 stems/ha)	1	ha	C.6.06	43.1	$1,832.33	$1,539.16	$25.98	$22	$15.00	13	$12.60	$469.71	$394.56	$2,343.03	$1,968			
A11.8	*IROD Laydown Area*																		$18,300		
A11.8.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	15	hrs	C.2.13	14.8	$44.39	$654.96	$194.68	$2,873	$67.01	1,030	$988.72	$0.00	$0.00	$306.07	$4,516			
A11.8.1.2			Re-grade slopes to be 3H:1V or flatter	0	hrs	C.2.13	0.0	$44.39	$0.00	$194.68	$0	$67.01	0	$0.00	$0.00	$0.00	$306.07	$0			
A11.8.1.3			Scarify surface	18,106	m2	C.2.16	0.0	$0.03	$629.15	$0.07	$1,189	$0.02	322	$308.88	$0.00	$0.00	$0.12	$2,127			
A11.8.2.1		Revegetate	Seed/Fertilize: broadcast seeding	2.33	ha	C.6.01	77.0	$1,029.50	$2,393.91	$401.72	$934	$232.94	564	$541.67	$1,005.72	$2,338.60	$2,669.89	$6,208			
A11.8.2.2			Tree seedling application (1,000 stems/ha)	2.33	ha	C.6.06	119.2	$1,832.33	$4,260.73	$25.98	$60	$15.00	36	$34.88	$469.71	$1,092.22	$2,343.03	$5,448			
A11.9	*Mill Area*																		$134,673		
A11.9.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	60	hrs	C.2.13	59.9	$44.39	$2,659.96	$194.68	$11,666	$67.01	4,183	$4,015.50	$0.00	$0.00	$306.07	$18,342			
A11.9.1.2			Scarify surface	42,365	m2	C.2.16	0.0	$0.03	$1,472.09	$0.07	$2,782	$0.02	753	$722.72	$0.00	$0.00	$0.12	$4,977			
A11.9.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	21,182	m3	R.055	384.7	$0.80	$16,942.37	$2.60	$55,057	$0.86	18,873	$18,118.30	$0.00	$0.00	$4.25	$90,118			
A11.9.3.1		Revegetate	Seed/Fertilize: broadcast seeding	4.24	ha	C.6.01	140.3	$1,029.50	$4,361.46	$401.72	$1,702	$232.94	1,028	$986.86	$1,005.72	$4,260.70	$2,669.89	$11,311			
A11.9.3.2			Tree seedling application (1,000 stems/ha)	4.24	ha	C.6.06	217.1	$1,832.33	$7,762.62	$25.98	$110	$15.00	66	$63.55	$469.71	$1,989.91	$2,343.03	$9,926			
A11.10	*Mill Valley Fill Extension (Stage 1 and 2)*																		$0		
A11.10.1		Costs included in Waste Rock Dump Section (WBS#: A1.5)		#VALUE!	hrs																
A11.11	*Minto South Portal*																		$87,165		
A11.11.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	51	hrs	C.2.13	51.5	$44.39	$2,284.70	$194.68	$10,020	$67.01	3,593	$3,449.00	$0.00	$0.00	$306.07	$15,754			
A11.11.1.2			Re-grade slopes to be 3H:1V or flatter	28	hrs	C.2.13	28.0	$44.39	$1,244.58	$194.68	$5,459	$67.01	1,957	$1,878.82	$0.00	$0.00	$306.07	$8,582			
A11.11.1.3			Scarify surface	62,765	m2	C.2.16	0.0	$0.03	$2,180.95	$0.07	$4,121	$0.02	1,115	$1,070.73	$0.00	$0.00	$0.12	$7,373			
A11.11.2.1		Revegetate	Seed/Fertilize: broadcast seeding	11.1	ha	C.6.01	366.5	$1,029.50	$11,389.08	$401.72	$4,444	$232.94	2,684	$2,576.99	$1,005.72	$11,125.98	$2,669.89	$29,536			
A11.11.2.2			Tree seedling application (1,000 stems/ha)	11.1	ha	C.6.06	567.0	$1,832.33	$20,270.56	$25.98	$287	$15.00	173	$165.94	$469.71	$5,196.26	$2,343.03	$25,920			
A11.12	*Pelly Laydown Area*																		$147,962		
A11.12.1.1		Re-grade	Flat areas: Re-grade to form tertiary drainage catchments	107	hrs	C.2.13	107.4	$44.39	$4,767.52	$194.68	$20,910	$67.01	7,497	$7,197.08	$0.00	$0.00	$306.07	$32,874			
A11.12.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	25,855	m3	R.058	380.7	$0.65	$16,779.44	$2.11	$54,545	$0.69	18,585	$17,841.31	$0.00	$0.00	$3.45	$89,165			
A11.12.3.1		Revegetate	Seed/Fertilize: broadcast seeding	5.17	ha	C.6.01	171.3	$1,029.50	$5,323.56	$401.72	$2,077	$232.94	1,255	$1,204.55	$1,005.72	$5,200.58	$2,669.89	$13,806			
A11.12.3.2			Tree seedling application (1,000 stems/ha)	5.17	ha	C.6.06	265.0	$1,832.33	$9,474.99	$25.98	$134	$15.00	81	$77.57	$469.71	$2,428.87	$2,343.03	$12,116			
A11.13	*W15 Sump Area Laydown*																		$36,022		
A11.13.1.1		Re-grade	Flat areas: Re-grade to ensure positive drainage	9	hrs	C.2.13	8.8	$44.39	$390.13	$194.68	$1,711	$67.01	613	$588.94	$0.00	$0.00	$306.07	$2,690			
A11.13.1.2			Slopes: Re-grade to 3H:1V	1	hrs	C.2.13	1.2	$44.39	$55.15	$194.68	$242	$67.01	87	$83.26	$0.00	$0.00	$306.07	$380			
A11.13.1.3			Scarify surface	9,484	m2	C.2.16	0.0	$0.03	$329.54	$0.07	$623	$0.02	169	$161.79	$0.00	$0.00	$0.12	$1,114			
A11.13.2.1		Cover	Load, haul, dump spread 0.5 m overburden cover	6,974	m3	R.065	106.1	$0.67	$4,675.54	$2.18	$15,199	$0.71	5,179	$4,971.43	$0.00	$0.00	$3.56	$24,846			
A11.13.3.1		Revegetate	Seed/Fertilize: broadcast seeding	1.39	ha	C.6.01	46.2	$1,029.50	$1,435.97	$401.72	$560	$232.94	338	$324.92	$1,005.72	$1,402.80	$2,669.89	$3,724			
A11.13.3.2			Tree seedling application (1,000 stems/ha)	1.39	ha	C.6.06	71.5	$1,832.33	$2,555.78	$25.98	$36	$15.00	22	$20.92	$469.71	$655.16	$2,343.03	$3,268			
A12	**Waste Disposal**																			**$452,923**	
A12.1	*Hydrocarbon contaminated soils*																		$66,551		
A12.1.1.1		Excavate HC contaminated	Complete testing for contaminated soils	48	test pits	C.3.03	24.0	$22.24	$1,067.53	$2.89	$139	$2.92	146	$139.97	$204.00	$9,792.00	$232.05	$11,138			
A12.1.1.2			Excavate and haul contaminated soils to on-site landfarm facility	726	m3	R.038	26.0	$1.58	$1,144.07	$3.17	$2,298	$0.96	723	$693.78	$0.00	$0.00	$5.70	$4,136			
A12.1.2.1		Construct landfarm	Existing Facility assumed to have sufficient capacity	0	m3																
A12.1.3.1		Operate landfarm	Aerate contaminated soils (mix)	12	event	C.3.04	270.0	$999.20	$11,990.37	$923.10	$11,077	$333.60	4,170	$4,003.20	$0.00	$0.00	$2,255.90	$27,071			
A12.1.3.2			Annual confirmation sampling	3	years	C.3.05	48.0	$711.69	$2,135.06	$92.63	$278	$93.31	292	$279.94	$1,224.00	$3,672.00	$2,121.63	$6,365			
A12.1.4.1		Close landfarm	Remove soils and collect liner	726	m3	R.001	22.9	$1.39	$1,007.57	$3.03	$2,197	$0.90	680	$652.50	$0.00	$0.00	$5.31	$3,857			
A12.1.4.2			Cut/fold liner and place into waste disposal container	4,265	m2	C.1.15	49.6	$0.43	$1,850.89	$0.19	$830	$0.07	308	$295.29	$0.00	$0.00	$0.70	$2,976			
A12.1.4.3			Regrade area	4,265	m2	C.2.11	42.7	$0.44	$1,893.14	$1.28	$5,472	$0.35	1,567	$1,504.69	$0.00	$0.00	$2.08	$8,870			
A12.1.4.4			Revegetation: Seed/Fertilizer: broadcast seeding	0.43	ha	C.6.01	14.1	$1,029.50	$439.08	$401.72	$171	$232.94	103	$99.35	$1,005.72	$428.94	$2,669.89	$1,139			
A12.1.4.5			Tree seedling application (1,000 stems/ha)	0.43	ha	C.6.06	21.9	$1,832.33	$781.49	$25.98	$11	$15.00	7	$6.40	$469.71	$200.33	$2,343.03	$999			
A12.2	*Metal Contaminated soils*																		$159,118		
A12.2.1.1		Crusher area	Delineate contaminated soil areas	197	test pits	C.3.01	98.5	$22.24	$4,381.32	$2.89	$570	$2.92	598	$574.45	$265.20	$52,244.40	$293.25	$57,770			
A12.2.1.2			Load, haul, dump contaminated soils to underground	2,949	m3	R.039	102.1	$1.53	$4,499.64	$3.33	$9,810	$0.99	3,035	$2,913.97	$0.00	$0.00	$5.84	$17,224			
A12.2.1.3			Complete confirmation testing	197	ea	C.3.02	32.8	$7.43	$1,463.48	$1.33	$261	$0.00	0	$0.00	$13.26	$2,612.22	$22.01	$4,337			
A12.2.2.1		Mill Area	Delineate contaminated soil areas	198	test pits	C.3.01	99.0	$22.24	$4,403.56	$2.89	$573	$2.92	601	$577.37	$265.20	$52,509.60	$293.25	$58,064			
A12.2.2.2			Load, haul, dump contaminated soils to underground	2,973	m3	R.039	103.0	$1.53	$4,536.49	$3.33	$9,890	$0.99	3,060	$2,937.83	$0.00	$0.00	$5.84	$17,365			
A12.2.2.3			Complete confirmation testing	198	ea	C.3.02	33.0	$7.43	$1,470.91	$1.33	$262	$0.00	0	$0.00	$13.26	$2,625.48	$22.01	$4,359			
A12.3	*Solid Waste Landfill*																		$217,134		
A12.3.1.1		Construction landfill	Place waste from site facilities	12,920	LCM	C.2.07	550.6	$1.89	$24,440.80	$8.30	$107,194	$2.86	38,433	$36,895.99	$0.00	$0.00	$13.04	$168,531			
A12.3.1.2			Place fill to minimize voids in the debris	3,371	m3	R.001	106.2	$1.39	$4,679.67	$3.03	$10,203	$0.90	3,157	$3,030.55	$0.00	$0.00	$5.31	$17,913			
A12.3.2.1		Close landfill	Soil Cover: Load haul, dump spread, compact (0.6m)	5,024	m3	R.002	190.0	$1.61	$8,100.50	$3.21	$16,144	$0.99	5,196	$4,988.33	$0.00	$0.00	$5.82	$29,233			
A12.3.2.2			Revegetation: Seed/Fertilizer: broadcast seeding	0.55	ha	C.6.01	18.1	$1,029.50	$562.06	$401.72	$219	$232.94	132	$127.18	$1,005.72	$549.07	$2,669.89	$1,458			
A12.4	*Hazardous Material Off-Site Disposal*																		$10,120		
A12.4.1.1		Hazardous Waste	From Building Demo: Transport off-site for disposal	2	trip	C.7.04	14.5	$351.24	$634.94	$303.64	$549	$113.33	213	$204.86	$0.00	$0.00	$768.21	$1,389			
A12.4.3.3			Hazardous materials: Disposal and Tipping fees.	1	ls	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A12.4.2.1		Reagents	Reagents: Transport to Whitehorse	25	hrs	C.1.22	99.3	$159.30	$3,955.90	$95.83	$2,380	$26.93	697	$668.71	$0.00	$0.00	$282.05	$7,004			
A12.4.2.2			Reagents: Disposal and tipping fees	2.2	trips	C.7.04	18.0	$351.24	$789.50	$303.64	$682	$113.33	265	$254.73	$0.00	$0.00	$768.21	$1,727			
A13	**Surface Water Conveyance**																			**$3,287,496**	
A13.1	*Channel A - W-15 to Main Pit*																		$207,036		
A13.1.1.1		Excavate channel	Load, haul, dump locally	40,300	m3	R.902	700.5	$0.77	$31,093.04	$1.64	$66,199	$0.46	19,351	$18,576.93	$0.00	$0.00	$2.88	$115,869			
A13.1.2.1		Place channel materials	Geotextile: Supply and place	7,785	m3	C.5.08	221.4	$1.09	$8,520.21	$0.11	$854	$0.11	897	$860.72	$2.16	$16,802.77	$3.47	$27,038			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	EOM Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A13.1.2.2			Rip-rap (angular, high quality): Screen and stockpile	3,855	m3	C.2.15	0.0	$2.54	$9,777.29	$6.06	$23,344	$1.86	7,462	$7,163.19	$0.00	$0.00	$10.45	$40,285			
A13.1.2.3			Rip-rap: Load, haul, dump	3,855	m3	R.018	102.2	$1.17	$4,501.31	$2.55	$9,814	$0.76	3,037	$2,915.04	$0.00	$0.00	$4.47	$17,230			
A13.1.2.4			Rip-rap: Place and secure	3,855	m3	R.904	41.9	$0.48	$1,860.68	$0.91	$3,507	$0.32	1,299	$1,247.51	$0.00	$0.00	$1.72	$6,615			
A13.2	*Channel B - W-35 to Area 2 Pit*																		$198,304		
A13.2.1.1		Excavate channel	Load, haul, dump locally	48,038	m3	R.902	835.0	$0.77	$37,063.29	$1.64	$78,910	$0.46	23,067	$22,143.92	$0.00	$0.00	$2.88	$138,117			
A13.2.2.1		Place channel materials	Geotextile: Supply and place	4,682	m2	C.5.08	133.1	$1.09	$5,124.06	$0.11	$514	$0.11	539	$517.64	$2.16	$10,105.20	$3.47	$16,261			
A13.2.2.2			Rip-rap (angular, high quality): Screen and stockpile	2,554	m3	C.2.15	0.0	$2.54	$6,477.46	$6.06	$15,465	$1.86	4,943	$4,745.61	$0.00	$0.00	$10.45	$26,689			
A13.2.2.3			Rip-rap: Load, haul, dump	2,554	m3	R.020	76.2	$1.32	$3,358.44	$2.87	$7,322	$0.85	2,266	$2,174.93	$0.00	$0.00	$5.03	$12,855			
A13.2.2.4			Rip-rap: Place and secure	2,554	m3	R.904	27.8	$0.48	$1,232.70	$0.91	$2,323	$0.32	861	$826.47	$0.00	$0.00	$1.72	$4,382			
A13.3	*Channel C - Main Pit to Main Access Road*																		$125,034		
A13.3.2.1		Excavate channel	Load, haul, dump locally	8,656	m3	R.902	150.5	$0.77	$6,678.68	$1.64	$14,219	$0.46	4,157	$3,990.26	$0.00	$0.00	$2.88	$24,888			
A13.3.3.1		Intake structure	Excavate soils to competent foundation materials	2,450	Bm3	R.902	42.6	$0.77	$1,890.07	$1.64	$4,024	$0.46	1,176	$1,129.25	$0.00	$0.00	$2.88	$7,043			
A13.3.3.2			Supply an install precast-concrete intake wingwall intake structure	1	ls	C.5.12	140.0	$5,946	$5,945.57	$3,583	$3,583	$643.84	671	$643.84	$37,000	$37,000.00	$47,172	$47,172			
A13.3.3.6			Backfill and compact around structure	2,450	Cm3	C.2.04	237.3	$2.58	$6,323.25	$3.01	$7,367	$1.03	2,638	$2,532.57	$0.00	$0.00	$6.62	$16,223			
A13.3.4.1		Place channel materials	Geotextile: Supply and place	3,171	m2	C.5.08	90.2	$1.09	$3,470.31	$0.11	$348	$0.11	365	$350.57	$2.16	$6,843.83	$3.47	$11,013			
A13.3.4.2			Rip-rap (angular, high quality): Screen and stockpile	1,037	m3	C.2.15	0.0	$2.54	$2,631.28	$6.06	$6,282	$1.86	2,008	$1,927.76	$0.00	$0.00	$10.45	$10,841			
A13.3.4.3			Rip-rap: Load, haul, dump	1,037	m3	R.022	36.4	$1.54	$1,600.27	$3.33	$3,452	$0.98	1,064	$1,021.09	$0.00	$0.00	$5.85	$6,073			
A13.3.4.4			Rip-rap: Place and secure	1,037	m3	R.904	11.3	$0.48	$500.75	$0.91	$944	$0.32	350	$335.73	$0.00	$0.00	$1.72	$1,780			
A13.4	*Channel D - Area 2 Pit to Channel E*																		$195,636		
A13.4.1.1		Excavate channel	Load, haul, dump locally	17,840	m3	R.902	310.1	$0.77	$13,763.89	$1.64	$29,304	$0.46	8,566	$8,223.41	$0.00	$0.00	$2.88	$51,291			
A13.4.2.1		Intake structure	Excavate soils to competent foundation materials	4,453	Bm3	R.902	77.4	$0.77	$3,435.48	$1.64	$7,314	$0.46	2,138	$2,052.57	$0.00	$0.00	$2.88	$12,802			
A13.4.2.2			Supply an install precast-concrete intake wingwall intake structure	1	ls	C.5.12	140.0	$5,946	$5,945.57	$3,583	$3,583	$643.84	671	$643.84	$37,000	$37,000.00	$47,172	$47,172			
A13.4.2.6			Backfill and compact around structure	4,453	Cm3	C.2.04	431.4	$2.58	$11,493.42	$3.01	$13,390	$1.03	4,795	$4,603.31	$0.00	$0.00	$6.62	$29,487			
A13.4.3.1		Place channel materials	Geotextile: Supply and place	4,910	m2	C.5.08	139.6	$1.09	$5,373.23	$0.11	$539	$0.11	565	$542.81	$2.16	$10,596.59	$3.47	$17,051			
A13.4.3.2			Rip-rap (angular, high quality): Screen and stockpile	2,199	m3	C.2.15	0.0	$2.54	$5,578.78	$6.06	$13,320	$1.86	4,258	$4,087.21	$0.00	$0.00	$10.45	$22,986			
A13.4.3.3			Rip-rap: Load, haul, dump	2,199	m3	R.020	65.7	$1.32	$2,892.50	$2.87	$6,306	$0.85	1,951	$1,873.18	$0.00	$0.00	$5.03	$11,072			
A13.4.3.4			Rip-rap: Place and secure	2,199	m3	R.904	23.9	$0.48	$1,061.68	$0.91	$2,001	$0.32	741	$711.81	$0.00	$0.00	$1.72	$3,774			
A13.5	*Channel E - Main Access Road to toe of MVFE*																		$739,690		
A13.5.1.1		Construct channel	Bulk fill to grade: Load, haul, dump, spread	29,750	m3																
A13.5.1.2			Excavate/form Channel: Load, haul, dump locally	46,386	m3	R.001	1,461.8	$1.39	$64,402.28	$3.03	$140,409	$0.90	43,445	$41,706.89	$0.00	$0.00	$5.31	$246,519			
A13.5.2.1		Place channel materials	Geotextile: Supply and place	27,307	m2	C.5.08	776.5	$1.09	$29,885.47	$0.11	$2,997	$0.11	3,145	$3,019.05	$2.16	$58,937.36	$3.47	$94,839			
A13.5.2.2			Rip-rap (angular, high quality): Screen and stockpile	20,558	m3	C.2.15	0.0	$2.54	$52,143.07	$6.06	$124,496	$1.86	39,794	$38,201.83	$0.00	$0.00	$10.45	$214,841			
A13.5.2.3			Rip-rap: Load, haul, dump	20,558	m3	R.025	845.7	$1.81	$37,226.38	$3.91	$80,296	$1.16	24,743	$23,753.12	$0.00	$0.00	$6.87	$141,275			
A13.5.2.4			Rip-rap: Place and secure	20,558	m3	R.904	223.6	$0.48	$9,923.17	$0.91	$18,701	$0.32	6,930	$6,653.05	$0.00	$0.00	$1.72	$35,277			
A13.5.3.1		Stilling basin/Energy Dissipa	Excavate basin (spoil locally)	645	m3	R.902	11.2	$0.77	$497.87	$1.64	$1,060	$0.46	310	$297.46	$0.00	$0.00	$2.88	$1,855			
A13.5.3.2			Geotextile: Supply and place	540	m2	C.5.08	15.4	$1.09	$591.07	$0.11	$59	$0.11	62	$59.71	$2.16	$1,165.65	$3.47	$1,876			
A13.5.3.3			Bedding layer: Screen and stockpile	131	m3	C.2.01	2.7	$0.96	$125.01	$2.11	$276	$0.44	60	$57.55	$0.00	$0.00	$3.51	$459			
A13.5.3.4			Bedding layer: Load, haul, dump and place	131	m3	R.028	7.7	$2.59	$337.69	$5.78	$755	$1.69	229	$220.12	$0.00	$0.00	$10.05	$1,312			
A13.5.3.5			Rip-rap (at inlet and outlet): Screen and stockpile	75	m3	C.2.15	0.0	$2.54	$190.23	$6.06	$454	$1.86	145	$139.37	$0.00	$0.00	$10.45	$784			
A13.5.3.6			Rip-rap (at inlet and outlet): Load, haul, dump	75	m3	R.027	3.2	$1.86	$139.15	$3.97	$298	$1.17	91	$87.84	$0.00	$0.00	$7.00	$525			
A13.5.3.7			Rip-rap (at inlet and outlet): Place and secure	75	m3	R.904	0.8	$0.48	$36.20	$0.91	$68	$0.32	25	$24.27	$0.00	$0.00	$1.72	$129			
A13.6	*Minto Creek Wetland By-Pass Channel*																		$231,721		
A13.6.1.1		Excavate channel	Load, haul, dump locally	29,668	m3	R.902	515.7	$0.77	$22,889.69	$1.64	$48,733	$0.46	14,246	$13,675.73	$0.00	$0.00	$2.88	$85,299			
A13.6.2.1		Place channel materials	Geotextile: Supply and place	12,212	m2	C.5.08	347.3	$1.09	$13,364.91	$0.11	$1,340	$0.11	1,406	$1,350.13	$2.16	$26,357.04	$3.47	$42,412			
A13.6.2.2			Rip-rap (angular, high quality): Screen and stockpile	5,950	m3	C.2.15	0.0	$2.54	$15,091.24	$6.06	$36,032	$1.86	11,517	$11,056.37	$0.00	$0.00	$10.45	$62,179			
A13.6.2.3			Rip-rap: Load, haul, dump	5,950	m3	R.001	187.5	$1.39	$8,260.73	$3.03	$18,010	$0.90	5,573	$5,349.64	$0.00	$0.00	$5.31	$31,620			Assumes WSP rip-rap can be re-used
A13.6.2.4			Rip-rap: Place and secure	5,950	m3	R.904	64.7	$0.48	$2,871.96	$0.91	$5,412	$0.32	2,006	$1,925.52	$0.00	$0.00	$1.72	$10,210			
A13.7	*Allowance: Minto North Spillway*																		$83,528		
A13.7.1.1		Excavate channel	Load, haul, dump locally	13,035	m3	R.902	226.6	$0.77	$10,056.98	$1.64	$21,412	$0.46	6,259	$6,008.67	$0.00	$0.00	$2.88	$37,477			
A13.7.2.1		Place channel materials	Geotextile: Supply and place	4,136	m2	C.5.08	117.6	$1.09	$4,526.14	$0.11	$454	$0.11	476	$457.23	$2.16	$8,926.04	$3.47	$14,363			
A13.7.2.2			Rip-rap (angular, high quality): Screen and stockpile	1,943	m3	C.2.15	0.0	$2.54	$4,929.10	$6.06	$11,769	$1.86	3,762	$3,611.23	$0.00	$0.00	$10.45	$20,309			
A13.7.2.3			Rip-rap: Load, haul, dump	1,943	m3	R.024	47.7	$1.08	$2,101.36	$2.36	$4,581	$0.70	1,418	$1,360.84	$0.00	$0.00	$4.14	$8,044			Assumes WSP rip-rap can be re-used
A13.7.2.4			Rip-rap: Place and secure	1,943	m3	R.904	21.1	$0.48	$938.04	$0.91	$1,768	$0.32	655	$628.91	$0.00	$0.00	$1.72	$3,335			
A13.8	*Ridgetop North Outlet Channel*																		$74,863		
A13.8.1.1		Excavate channel	Load, haul, dump locally	7,758	m3	R.902	134.9	$0.77	$5,985.73	$1.64	$12,744	$0.46	3,725	$3,576.25	$0.00	$0.00	$2.88	$22,306			
A13.8.2.1		Place channel materials	Geotextile: Supply and place	3,995	m2	C.5.08	113.6	$1.09	$4,371.69	$0.11	$438	$0.11	460	$441.63	$2.16	$8,621.43	$3.47	$13,873			
A13.8.2.2			Rip-rap (angular, high quality): Screen and stockpile	2,531	m3	C.2.15	0.0	$2.54	$6,419.56	$6.06	$15,327	$1.86	4,899	$4,703.20	$0.00	$0.00	$10.45	$26,450			
A13.8.2.3			Rip-rap: Load, haul, dump	2,531	m3	R.068	33.1	$0.58	$1,455.99	$1.90	$4,821	$0.64	1,681	$1,613.82	$0.00	$0.00	$3.12	$7,891			Assumes WSP rip-rap can be re-used
A13.8.2.4			Rip-rap: Place and secure	2,531	m3	R.904	27.5	$0.48	$1,221.69	$0.91	$2,302	$0.32	853	$819.09	$0.00	$0.00	$1.72	$4,343			
A13.9	*Secondary Catchment Channels*																		$791,748		
A13.9.1.1		Southwest Dump	Excavate mild-graded channels (good access), spoil locally	10,787	Bm3	R.902	187.5	$0.77	$8,322.78	$1.64	$17,720	$0.46	5,180	$4,972.55	$0.00	$0.00	$2.88	$31,015			
A13.9.1.2			Excavate steep-graded channels (poor access), spoil locally	6,640	Bm3	R.903	131.0	$0.88	$5,815.42	$1.86	$12,381	$0.52	3,619	$3,474.49	$0.00	$0.00	$3.26	$21,671			
A13.9.1.3			Geotextile: Supply and place	25,206	m2	C.5.08	716.7	$1.09	$27,585.63	$0.11	$2,766	$0.11	2,903	$2,786.72	$2.16	$54,401.82	$3.47	$87,541			
A13.9.1.4			Bedding layer: Screen and stockpile	2,054	Cm3	C.2.01	41.8	$0.96	$1,967.27	$2.11	$4,343	$0.44	943	$905.70	$0.00	$0.00	$3.51	$7,216			
A13.9.1.5			Bedding layer: Load, haul, dump and place in steep areas	2,054	Cm3	R.032	72.2	$1.55	$3,176.63	$3.34	$6,852	$0.99	2,111	$2,026.92	$0.00	$0.00	$5.87	$12,055			
A13.9.1.6			Rip-rap (angular, high quality): Screen and stockpile	6,806	Cm3	C.2.15	0.0	$2.54	$17,263.99	$6.06	$41,219	$1.86	13,175	$12,648.20	$0.00	$0.00	$10.45	$71,131			
A13.9.1.7			Rip-rap: Load, haul, dump	6,806	m3	R.031	223.5	$1.45	$9,846.77	$3.15	$21,468	$0.94	6,642	$6,376.76	$0.00	$0.00	$5.54	$37,691			
A13.9.1.8			Rip-rap: Place in mild-graded channels (good access)	3,787	m3	R.904	41.2	$0.48	$1,828.10	$0.91	$3,445	$0.32	1,277	$1,225.66	$0.00	$0.00	$1.72	$6,499			
A13.9.1.9			Rip-rap: Place in steep-graded channels (poor access)	3,019	m3	R.905	74.5	$1.10	$3,308.57	$2.33	$7,044	$0.65	2,059	$1,976.75	$0.00	$0.00	$4.08	$12,329			
A13.9.2.1		Main Waste Dump	Excavate mild-graded channels (good access), spoil locally	6,477	Bm3	R.902	112.6	$0.77	$4,997.00	$1.64	$10,639	$0.46	3,110	$2,985.52	$0.00	$0.00	$2.88	$18,621			
A13.9.2.2			Excavate steep-graded channels (poor access), spoil locally	3,986	Bm3	R.903	78.6	$0.88	$3,490.67	$1.86	$7,432	$0.52	2,172	$2,085.55	$0.00	$0.00	$3.26	$13,008			
A13.9.2.3			Geotextile: Supply and place	15,132	m2	C.5.08	430.3	$1.09	$16,560.93	$0.11	$1,661	$0.11	1,743	$1,672.99	$2.16	$32,659.93	$3.47	$52,555			
A13.9.2.4			Bedding layer: Screen and stockpile	1,233	Cm3	C.2.01	25.1	$0.96	$1,180.84	$2.11	$2,607	$0.44	566	$543.64	$0.00	$0.00	$3.51	$4,331			
A13.9.2.5			Bedding layer: Load, haul, dump and place in steep areas	1,233	Cm3	R.030	34.8	$1.25	$1,535.65	$2.76	$3,401	$0.82	1,059	$1,016.39	$0.00	$0.00	$4.83	$5,953			
A13.9.2.6			Rip-rap (angular, high quality): Screen and stockpile	4,086	Cm3	C.2.15	0.0	$2.54	$10,364.11	$6.06	$24,745	$1.86	7,909	$7,593.11	$0.00	$0.00	$10.45	$42,702			
A13.9.2.7			Rip-rap: Load, haul, dump	4,086	m3	R.029	101.4	$1.09	$4,473.09	$2.42	$9,908	$0.72	3,084	$2,960.56	$0.00	$0.00	$4.24	$17,341			
A13.9.2.8			Rip-rap: Place in mild-graded channels (good access)	2,274	m3	R.904	24.7	$0.48	$1,097.59	$0.91	$2,068	$0.32	767	$735.89	$0.00	$0.00	$1.72	$3,902			
A13.9.2.9			Rip-rap: Place in steep-graded channels (poor access)	1,812	m3	R.905	44.7	$1.10	$1,985.96	$2.33	$4,228	$0.65	1,236	$1,186.53	$0.00	$0.00	$4.08	$7,401			
A13.9.3.1		Reclamation Overburden Di	Excavate mild-graded channels (good access), spoil locally	4,886	Bm3	R.902	84.9	$0.77	$3,769.89	$1.64	$8,026	$0.46	2,346	$2,252.36	$0.00	$0.00	$2.88	$14,049			
A13.9.3.2			Excavate steep-graded channels (poor access), spoil locally	3,007	Bm3	R.903	59.3	$0.88	$2,633.47	$1.86	$5,607	$0.52	1,639	$1,573.40	$0.00	$0.00	$3.26	$9,814			
A13.9.3.3			Geotextile: Supply and place	11,416	m2	C.5.08	324.6	$1.09	$12,494.07	$0.11	$1,253	$0.11	1,315	$1,262.16	$2.16	$24,639.65	$3.47	$39,649			
A13.9.3.4			Bedding layer: Screen and stockpile	930	Cm3	C.2.01	18.9	$0.96	$890.86	$2.11	$1,967	$0.44	427	$410.14	$0.00	$0.00	$3.51	$3,268			
A13.9.3.5			Bedding layer: Load, haul, dump and place in steep areas	930	Cm3	R.034	34.7	$1.64	$1,527.43	$3.58	$3,330	$1.06	1,030	$989.17	$0.00	$0.00	$6.29	$5,847			
A13.9.3.6			Rip-rap (angular, high quality): Screen and stockpile	3,083	Cm3	C.2.15	0.0	$2.54	$7,819.00	$6.06	$18,669	$1.86	5,967	$5,728.47	$0.00	$0.00	$10.45	$32,216			
A13.9.3.7			Rip-rap: Load, haul, dump	3,083	m3	R.033	97.0	$1.39	$4,271.96	$3.02	$9,314	$0.90	2,882	$2,766.52	$0.00	$0.00	$5.30	$16,352			
A13.9.3.8			Rip-rap: Place in mild-graded channels (good access)	1,715	m3	R.904	18.7	$0.48	$828.06	$0.91	$1,561	$0.32	578	$555.18	$0.00	$0.00	$1.72	$2,944			
A13.9.3.9			Rip-rap: Place in steep-graded channels (poor access)	1,367	m3	R.905	33.8	$1.10	$1,498.27	$2.33	$3,190	$0.65	932	$895.16	$0.00	$0.00	$4.08	$5,583			
A13.9.4.1		DSTSF & MVFE	Excavate mild-graded channels (good access), spoil locally	7,378	Bm3	R.902	128.3	$0.77	$5,692.78	$1.64	$12,120	$0.46	3,543	$3,401.22	$0.00	$0.00	$2.88	$21,214			
A13.9.4.2			Excavate steep-graded channels (poor access), spoil locally	4,541	Bm3	R.903	89.6	$0.88	$3,976.72	$1.86	$8,467	$0.52	2,475	$2,375.94	$0.00	$0.00	$3.26	$14,819			
A13.9.4.3			Geotextile: Supply and place	17,239	m2	C.5.08	490.2	$1.09	$18,866.88	$0.11	$1,892	$0.11	1,985	$1,905.94	$2.16	$37,207.52	$3.47	$59,872			
A13.9.4.4			Bedding layer: Screen and stockpile	1,405	Cm3	C.2.01	28.6	$0.96	$1,345.27	$2.11	$2,970	$0.44	645	$619.34	$0.00	$0.00	$3.51	$4,935			
A13.9.4.5			Bedding layer: Load, haul, dump and place in steep areas	1,405	Cm3	R.028	82.5	$2.59	$3,633.88	$5.78	$8,121	$1.69	2,467	$2,368.79	$0.00	$0.00	$10.05	$14,124			

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	EOM Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
A13.9.4.6			Rip-rap (angular, high quality): Screen and stockpile	4,655	Cm3	C.2.15	0.0	$2.54	$11,807.22	$6.06	$28,191	$1.86	9,011	$8,650.38	$0.00	$0.00	$10.45	$48,648			
A13.9.4.7			Rip-rap: Load, haul, dump	4,655	m3	R.027	196.3	$1.86	$8,636.41	$3.97	$18,485	$1.17	5,679	$5,451.80	$0.00	$0.00	$7.00	$32,573			
A13.9.4.8			Rip-rap: Place in mild-graded channels (good access)	2,590	m3	R.904	28.2	$0.48	$1,250.42	$0.91	$2,356	$0.32	873	$838.35	$0.00	$0.00	$1.72	$4,445			
A13.9.4.9			Rip-rap: Place in steep-graded channels (poor access)	2,065	m3	R.905	51.0	$1.10	$2,262.48	$2.33	$4,817	$0.65	1,408	$1,351.75	$0.00	$0.00	$4.08	$8,431			
A13.9.5.1		Ridgetop Waste Dump	Excavate mild-graded channels (good access), spoil locally	574	Bm3	R.902	10.0	$0.77	$442.77	$1.64	$943	$0.46	276	$264.54	$0.00	$0.00	$2.88	$1,650			
A13.9.5.2			Excavate steep-graded channels (poor access), spoil locally	353	Bm3	R.903	7.0	$0.88	$309.30	$1.86	$659	$0.52	192	$184.80	$0.00	$0.00	$3.26	$1,153			
A13.9.5.3			Geotextile: Supply and place	876	m2	C.5.08	24.9	$1.09	$958.89	$0.11	$96	$0.11	101	$96.87	$2.16	$1,891.03	$3.47	$3,043			
A13.9.5.4			Bedding layer: Screen and stockpile	465	Cm3	C.2.01	9.4	$0.96	$444.99	$2.11	$982	$0.44	213	$204.86	$0.00	$0.00	$3.51	$1,632			
A13.9.5.5			Bedding layer: Load, haul, dump and place in steep areas	109	Cm3	R.063	2.9	$1.18	$128.79	$3.83	$419	$1.26	143	$137.73	$0.00	$0.00	$6.27	$685			
A13.9.5.6			Rip-rap (angular, high quality): Screen and stockpile	362	Cm3	C.2.15	0.0	$2.54	$918.34	$6.06	$2,193	$1.86	701	$672.81	$0.00	$0.00	$10.45	$3,784			
A13.9.5.7			Rip-rap: Load, haul, dump	362	m3	R.064	8.5	$1.03	$373.01	$3.40	$1,231	$1.14	431	$413.67	$0.00	$0.00	$5.57	$2,018			
A13.9.5.8			Rip-rap: Place in mild-graded channels (good access)	201	m3	R.904	2.2	$0.48	$97.26	$0.91	$183	$0.32	68	$65.21	$0.00	$0.00	$1.72	$346			
A13.9.5.9			Rip-rap: Place in steep-graded channels (poor access)	161	m3	R.905	4.0	$1.10	$175.97	$2.33	$375	$0.65	110	$105.14	$0.00	$0.00	$4.08	$656			
A13.10	*Tailings Diversion Ditch*																		$624,970		
A13.10.1.2		Regrade	Slopes: Re-grade to 3H:1V	28	hrs	C.2.13	28	$44.39	$1,226	$194.68	$5,376	$67.01	1,928	$1,850	$0.00	$0	$306.07	$8,452			
A13.10.2.1		Cover	Flat area cover material: Load, haul, dump spread	25,182	Cm3	R.066	562.3	$0.98	$24,763	$3.20	$80,473	$1.05	27,585	$26,482	$0.00	$0	$5.23	$131,718			
A13.10.2.2			Slope area cover material: Load, haul, dump along crest	18,049	Cm3	R.067	352.6	$0.86	$15,513	$2.84	$51,216	$0.95	17,921	$17,204	$0.00	$0	$4.65	$83,933			
A13.10.2.3			Slope area cover material: Spread down slope	14	hrs	C.2.13	14.0	$44.39	$621	$194.68	$2,724	$67.01	977	$937	$0.00	$0	$306.07	$4,282			
A13.10.3.1		Revegetate	Seed/Fertilize: tractor application	9	ha	C.6.01	286.4	$1,029.50	$8,901	$401.72	$3,473	$232.94	2,098	$2,014	$1,005.72	$8,696	$2,670	$23,085			
A13.10.3.2			Tree seedling application (1,000 stems/ha)	9	ha	C.6.06	443.1	$1,832.33	$15,843	$25.98	$225	$15.00	135	$130	$469.71	$4,061	$2,343	$20,258			
A13.10.4.1		Extend Channel to WSP	Clear and grub footprint	13,889	m2	C.2.03	52.8	$0.17	$2,344	$0.49	$6,774	$0.13	1,940	$1,863	$0.00	$0	$0.79	$10,981			
A13.10.4.2			Geotextile: Supply and place	13,889	m2	C.5.08	394.9	$1.09	$15,200	$0.11	$1,524	$0.11	1,600	$1,536	$2.16	$29,976	$3.47	$48,236			
A13.10.4.3			Rip-rap (angular, high quality): Screen and stockpile	13,169	m3	C.2.15	0.0	$2.54	$33,403	$6.06	$79,753	$1.86	25,492	$24,472	$0.00	$0	$10.45	$137,629			
A13.10.4.4			Rip-rap: Load, haul, dump	13,169	m3	R.035	618.4	$2.07	$27,207	$4.42	$58,233	$1.30	17,890	$17,175	$0.00	$0	$7.79	$102,615			
A13.10.4.5			Rip-rap: Place in steep-graded channels (poor access)	13,169	m3	R.905	325.1	$1.10	$14,432	$2.33	$30,726	$0.65	8,982	$8,622	$0.00	$0	$4.08	$53,780			
A14	**Water Treatment**																			**$483,085**	
A14.1	*Passive Treatment System - Minto Creek Valley*																		$397,994		
A14.1.1.1		Water Management	Install pump and pump around system	8	hrs	C.4.14	32.0	$120.43	$963.46	$20.53	$164	$6.80	57	$54.40	$0.00	$0.00	$147.76	$1,182			Assumed
A14.1.1.2			Maintain pump around system during wetland construction activities	1,440	hrs	C.4.13	144.0	$3.55	$5,117.32	$12.98	$18,691	$2.00	3,000	$2,880.00	$0.00	$0.00	$18.53	$26,689			Assumes wetlands constructed over 2 months
A14.1.2.1		Valley Fill placement	Load, haul, dump material from the WSP excavation	85,918	m3	n/a - costed	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A14.1.2.1			Load, haul dump valley fill material	0	m3	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			Fill from WSP excavation exceeds the required amount.
A14.1.3.1		Construct wetlands	Excavate cells: Load, haul dump, spoil locally	14,459	m3	R.902	251.3	$0.77	$11,155.61	$1.64	$23,751	$0.46	6,943	$6,665.06	$0.00	$0.00	$2.88	$41,572			
A14.1.3.2			Subgrade preparation: remove sharp objects, place fill as required	9,273	m2	C.2.18	317.0	$1.31	$12,142.93	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$1.31	$12,143			
A14.1.3.3			Geotextile: Supply and install	10,201	m2	C.5.08	290.1	$1.09	$11,163.84	$0.11	$1,120	$0.11	1,175	$1,127.78	$2.16	$22,016.30	$3.47	$35,428			
A14.1.3.4			HDPE liner: Supply and install	10,201	m2	C.5.11	1,187.0	$4.74	$48,319.12	$0.27	$2,749	$0.27	2,884	$2,769.07	$9.38	$95,723.03	$14.66	$149,560			
A14.1.3.5			Organics: Mulch organics, produce and stockpile	5,122	Cm3	C.2.10	577.1	$4.50	$23,023.50	$4.87	$24,922	$1.82	9,720	$9,331.25	$0.00	$0.00	$11.18	$57,277			
A14.1.3.6			Organics: Load, haul, dump, place in cell	5,122	Cm3	R.001	161.4	$1.39	$7,111.43	$3.03	$15,504	$0.90	4,797	$4,605.36	$0.00	$0.00	$5.31	$27,221			
A14.1.3.7			Plant wetland vegetation	0.9	ha	C.6.05	40.6	$1,443.27	$1,338.39	$307.03	$285	$180.00	174	$166.92	$1,159.23	$1,074.99	$3,089.53	$2,865			
A14.1.4.1		Conveyance channels	Excavate channels: Load, haul, dump locally	1,402	m3	R.902	24.4	$0.77	$1,081.90	$1.64	$2,303	$0.46	673	$646.39	$0.00	$0.00	$2.88	$4,032			
A14.1.4.2			Surface preparation: remove sharp objects, place fill as required	1,858	m2	C.2.18	63.5	$1.31	$2,432.63	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$1.31	$2,433			
A14.1.4.3			Geotextile: Supply and place	1,858	m2	C.5.08	52.8	$1.09	$2,033.17	$0.11	$204	$0.11	214	$205.39	$2.16	$4,009.63	$3.47	$6,452			
A14.1.4.4			HDPE Liner: Supply and install	1,858	m2	C.5.11	216.2	$4.74	$8,799.92	$0.27	$501	$0.27	525	$504.31	$9.38	$17,433.16	$14.66	$27,238			
A14.1.4.5			Bedding/Protection layer: Screen and stockpile	442	m3	C.2.01	9.0	$0.96	$423.54	$2.11	$935	$0.44	203	$194.99	$0.00	$0.00	$3.51	$1,554			
A14.1.4.6			Bedding/Protection layer: Load, haul, dump and place	442	m3	R.001	13.9	$1.39	$614.04	$3.03	$1,339	$0.90	414	$397.65	$0.00	$0.00	$5.31	$2,350			Assumes WSP fill can be re-used
A14.1.5.1		Operation and maintenance	System to be started Yr. 1 post-closure	0	yr	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A14.2	*Southwest Wetlands*																		$15,944		
A14.2.1		Berm construction	Construct 3 berms across valley	3,000	m3	R.001	94.5	$1.39	$4,165.24	$3.03	$9,081	$0.90	2,810	$2,697.41	$0.00	$0.00	$5.31	$15,944			Assumed
A14.2.2		Outlet structures	Excavate outlet channels through berms	1	ls																Allowance
A14.3	*Decommission/demolish Water Treatment System*																		$69,147		
A14.3.1.1		Remove equipment	Small equipment: dismantle and prep for transport	17	hrs	C.1.13	67.1	$168.00	$2,819.22	$86.19	$1,446	$14.53	254	$243.80	$0.00	$0.00	$268.71	$4,509			
A14.3.1.2			Large equipment (crane req'd): dismantle and prep for transport	29	hrs	C.1.14	175.7	$269.89	$7,903.93	$167.56	$4,907	$20.53	626	$601.18	$0.00	$0.00	$457.98	$13,412			
A14.3.2.1		Prepare for demolition	Reagents: Load for transport	25	hrs	C.1.22	99.3	$159.30	$3,955.90	$95.83	$2,380	$26.93	697	$668.71	$0.00	$0.00	$282.05	$7,004			
A14.3.2.2			Reagents: Transport to Whitehorse	2.2	trips	C.7.04	18.0	$351.24	$789.50	$303.64	$682	$113.33	265	$254.73	$0.00	$0.00	$768.21	$1,727			
A14.3.2.3			Reagents: Disposal and tipping fees	1.0	ls	n/a	0.0	$0.00	$0.00	$0.00	$0	$0.00	0	$0.00	$0.00	$0.00	$0.00	$0			
A14.3.3.1		Demolition	Structural building demolition	163	tonnes	C.1.05	325.0	$69.36	$11,272.12	$139.08	$22,604	$33.84	5,729	$5,499.69	$0.00	$0.00	$242.28	$39,376			
A14.3.3.2			Other demolition: miscellaneous	15	m3	C.1.07	1.5	$3.74	$56.09	$2.07	$31	$0.74	12	$11.05	$0.00	$0.00	$6.55	$98			
A14.3.4.1		Waste disposal	Off-site disposal (re-usable equipment, etc.)	3	trips	C.7.07	27.1	$355.10	$1,202.32	$423.68	$1,435	$113.33	400	$383.71	$0.00	$0.00	$892	$3,021			
Subtotal Direct Costs - Active Closure							**115,225**		**$4,898,304**		**$8,438,294**			**$2,557,216**		**$1,093,950**				**$16,987,765**	

							Labour			Equipment		Fuel			Material		Activity Totals		Subtotals		
WBS	Facility/Area	Task	Activity	EOM Qnty	Units	Cost Code	Total Mhrs	Unit Rate	Cost	Unit Rate	Cost	Unit Rate	Consumed (L)	Cost	Unit Rate	Cost	Unit Rate	Cost	WBS Level 2	WBS Level 1	Source / Comments
INDIRECT COSTS																					
B1	**Mobilization-Demobilization**																			**$324,077**	
B1.1		Mobilization	Year 0	1	LS												$162,039	$162,038.53			
B1.2		Demobilization	Year 3 (end of Active Closure)	1	LS												$144,406	$144,405.50			
B1.4		Demobilization	End of Passive closure	1	LS												$17,633	$17,633.03			
B2	**Transportation Costs**																			**$419,142**	
B2.1.1		Barge Operations		15	month												$10,832	$162,486			
		Staffing Bus trips during barge operation period		60	ea (one per week)												$678	$40,656			Turnaround labour costs included in labour build-up
B2.2.2		Air transport and airstrip operations		72	flights												$3,000	$216,000			Three flights per week
B3	**Site/Road Maintenance**																			**$791,662**	
B3.1	*Road Maintenance*																		$727,887		
B3.1.1		Water Truck	Assume needed 150 hrs per month	21	month												$12,153	$255,205			
B3.1.2		Grader	Assume needed 150 hrs per month (16H + Operator)	21	month												$22,509	$472,682			Includes Main Access Road throughout closure period.
B3.2	*Soil Erosion*																		$63,775		
B3.2.1		Erosion control allowance	Silt fencing: supply, install remove at strategic locations	7024	m	C.2.17	283.5	$3.23	$22,661.96	$1.93	$13,533	$0.68	4,941	$4,743.08	$0.92	$6,485.58	$6.75	$47,424			
B3.2.2			Supply and install erosion control matting	10000	m2	C.2.09	175.4	$0.66	$6,622.75	$0.09	$912	$0.05	548	$526.32	$0.83	$8,290.76	$1.64	$16,352				
B4	**Construction Support**																			**$5,263,751**	
B4.1	*Field Support Staff*																		$3,235,218		Turnaround labour costs included in labour build-up
B4.1.1		Mine Manager		21	month			(360) site working hours per month									$23,408	$491,578			
B4.1.2		Office/Camp manager		21	month			(360) site working hours per month									$22,267	$467,614			
B4.1.3		Security/Administrative Assistant		21	month			(360) site working hours per month									$16,600	$348,591			
B4.1.4		Foreman		21	month			(360) site working hours per month									$17,769	$373,152			
B4.1.5		Mechanic		21	month			(360) site working hours per month									$17,769	$373,152			
B4.1.6		Surveyor		21	month			(360) site working hours per month									$17,769	$373,152			
B4.1.7		Engineering technician	Material & QA/QC testing	10.5	month			(360) site working hours per month									$42,120	$442,260			Assumed needed 1/2 of time
B4.1.8		Medic/H&S supervisor		21	month			(360) site working hours per month									$17,415	$365,718			
B4.2	*Field support Vehicles*																		$862,785		
B4.2.1		Pick-up trucks (4 required)		84	month												$4,446	$373,452			
B4.2.2		Fuel truck		21	month												$4,720	$99,123			
B4.2.3		Mechanic service vehicle		21	month												$11,803	$247,873			
B4.2.4		Emergency transport vehicle		21	month												$4,519	$94,901			
B4.2.5		Passenger bus		21	month												$2,259	$47,436			
B4.3	*Field Support Equipment/Supplies*																		$1,165,748		
B4.3.1		Light Towers		21	month			2 # of Units									$3,356	$70,476			
B4.3.2		Material/Laboratory testing allowance		21	month												$1,000	$21,000			
B4.4		Construction, Environmental, and H&S Management Plans		1	LS												$25,000	$25,000			
B4.5		Office supplies		21	month												$1,000	$21,000			
B4.6		Communications		21	month												$1,000	$21,000			
B4.7		Misc. supplies		21	month												$500	$10,500			
B4.8		Camp costs	Includes catering and housekeeping	11,523	man-day												$80	$921,801			
B4.9		Power and heat		21	month												$3,570	$74,970			
B5	**QA and Project Management**																			**$1,752,126**	
B5.1		Project Manager		21	month			(160) site working hours per month									$9,897	$207,828			
B5.2		Design Engineer		21	month			(360) site working hours per month									$48,600	$1,020,600			
B5.3		Environmental Monitor		21	month			(360) site working hours per month									$16,046	$336,972			
B5.4		Engineering , Design, and Construction plans (included in Planning and Permitting Costs)															$0.00	$0			
B5.5		Field support vehicles (2 vehicles)		42	month												$4,446	$186,726			
Subtotal Indirect Costs - Active Closure																				**$8,550,758**	
CLOSURE IMPLEMENTATION COSTS - TOTAL																				**$25,538,523**	

Worksheet 15 - End-of-Mine Liability Estimate - Schedule Details

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

A: Year 0 Site Access Schedule

Stage	Flight Months	Helicopter Months	Barge Months	Total Months	Month: Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	Comments/Notes
Interim Care	0	0	0	0													
SUB-TOTAL INTERIM CARE	**0**	**0**	**0**	**0**													
Active Closure 1	2	0	5	7				F	F	B	B	B	B	B			
Active Closure 2	2	0	5	7				F	F	B	B	B	B	B			
Active Closure 3	2	0	5	7				F	F	B	B	B	B	B			
SUB-TOTAL ACTIVE CLOSURE	**6**	**0**	**15**	**21**													
Post-Closure 1	2	0	3	5				F	F	B	B	B					
SUB-TOTAL POST-CLOSURE 1	**2**	**0**	**3**	**5**													
Post-Closure 2	0	0	1	1						B							
SUB-TOTAL POST-CLOSURE 2	**0**	**0**	**1**	**1**													

B: Annual C&M Staffing Schedule - Interim Operations

No Interim operations included under the EOM scenario.

C: Annual C&M Staffing Schedule - During Active Phase

#	Role		Months Required	Month: Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	Comments/Notes
Months Active at Site			**7**				**1**	**1**	**1**	**1**	**1**	**1**	**1**			
Water Treatment/Environmental Staffing																
1	Warehouse/Water Treatment Operator		5				1.0	1.0	1.0	1.0	1.0					
2	Environmental Manager		4				1.0	0.5	0.5	0.5	0.5	0.5	0.5			
3	Environmental Technician		4				1.0	0.5	0.5	0.5	0.5	0.5	0.5			
TOTAL			**13**	**0**	**0**	**0**	**3**	**2**	**2**	**2**	**2**	**1**	**1**	**0**	**0**	
	Annual man-days:		**390**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)

D: Annual C&M Staffing Schedule - Post Closure 1

#	Role		Months Required	Month: Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	Comments/Notes
Months Active at Site			**5**				**1**	**1**	**1**	**1**	**1**					
Administration/Office																
1	Mine Manager		3				0.5	0.5	0.5	0.5	0.5					
2	Office/Camp Mgr		3				1.0	0.5	0.5	0.5	0.5					
3	Payroll/Accounting/HR		1				0.25	0.25	0.25	0.25	0.25					
Water Treatment/Environmental Staffing																
1	Water Treatment Operator		2				0.5	0.5	0.3	0.3	0.3					
2	Environmental Manager		3				0.5	0.5	0.5	0.5	0.5					
3	Environmental Technician		3				0.5	0.5	0.5	0.5	0.5					
Operations																
4	Barge Operator		6						2.0	2.0	2.0					
5	Equipment operators		10				2.0	3.0	1.5	1.5	1.5					
6	HD Mechanic		2				1.0	0.25	0.25	0.25	0.25					
7	Tradesmen		2				1.0	0.25	0.25	0.25	0.25					
8	Labour/Helpers		2				1.0	0.25	0.25	0.25	0.5					
Camp/Support																
1	Cooks/Housecleaning etc.		0				0.0	0.0	0.0	0.0	0.0					Contract (accounted for in camp costs)
Other																
1	Visitors		1				0.25				0.25					
TOTAL			**36**	**0**	**0**	**0**	**9**	**7**	**7**	**7**	**7**	**0**	**0**	**0**	**0**	
	Annual man-days:		**1,073**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)
2. Orange highlighted cells are used for camp-man day calculations only.

E: Annual C&M Staffing Schedule - Post Closure 2

				Month												
#	Role		Months Required	Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	Comments/Notes
Months Active at Site			**1**						**1**							
Administration/Office																
1	Mine Manager		1						0.5							
2	Office/Camp Mgr		0						0.0							
3	Payroll/Accounting/HR		0						0.00							
Water Treatment/Environmental Staffing																
1	Passive Treatment Scientist		1						0.5							
2	Environmental Manager		0													
3	Environmental Technician		0													
Operations																
4	Barge Operator		1						1.0							
5	Equipment operators		1						1.0							
6	HD Mechanic		1						0.50							
7	Tradesmen		0													
8	Labour/Helpers		0						0.25							
Camp/Support																
1	Cooks/Housecleaning etc.		0						0.0							Contract (accounted for in camp costs)
Other																
1	Visitors		0						0.25							
TOTAL			**4**	**0**	**0**	**0**	**0**	**0**	**4**	**0**	**0**	**0**	**0**	**0**	**0**	
	Annual man-days:		**120**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)
2. Orange highlighted cells are used for camp-man day calculations only.

E: Annual C&M Staffing Schedule - Long-term perpetual maintenance year

				Month												
#	Role		Months Required	Jan	Feb	Mar	Apr	May	Jun	Jul	Aug	Sep	Oct	Nov	Dec	Comments/Notes
Months Active at Site			**1**							**1**						
Administration/Office																
1	Project Manager/Foreman		1							1.0						
Water Treatment/Engineering Staffing																
1	Passive Treatment Scientist		1							1.0						
2	Site Engineer		1							1.0						
3	Environmental Technician		0													
Operations																
4	Barge Operator		1							1.0						Assumed required 1/2 the time (near mob and demob)
5	Equipment operators		4							4.0						
6	HD Mechanic		1							0.50						
7	Tradesmen		0													
8	Labour/Helpers		1							1.00						
Camp/Support																
1	Cooks/Housecleaning etc.		0													Contract (accounted for in camp costs)
Other																
1	Visitors		1							0.50						
TOTAL			**10**	**0**	**0**	**0**	**0**	**0**	**0**	**10**	**0**	**0**	**0**	**0**	**0**	
	Annual man-days:		**300**													

Notes:
1. 0.5 personnel indicates staff on site 50% of time (i.e. no cross shift)
2. Orange highlighted cells are used for camp-man day calculations only.

Worksheet 16 - End-of-Mine Liability Estimate - Planning and Permitting Costs

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

Planning And Permitting

WBS	Facility/Area	Task	Activity	Qnty	Unit	Total Unit Rate ($/unit)	Activity Total	Subtotal	Source / Comments
M1	**Planning and Permitting**								
M1.1	*Reclamation Research/Planning*							$200,000	
M1.1.1		Complete reclamation closure and research plan		1	yr	$200,000	$200,000		
M1.2	*Technical studies and investigations*							$58,500	
M1.2.1		Allowance for additional technical studies		1	ls	$58,500	$58,500		
M1.3	*Monitoring and Management Plans*							$209,158	
M1.3.1		Adaptive Mgmt Plans	Physcial, water quality, etc	1	ls	$15,000	$15,000		
M1.3.2		Revegetation plan		1	ls	$30,000	$30,000		
M1.3.3		Waste Management Plan	Water treatment, sludge, landfarm, etc.	1	ls	$50,000	$50,000		
M1.3.4		MMER Final Monitoring Study		1	ls	$114,158	$114,158		Quote: Minnow
M1.4	*Engineering, Design, and Construction Plans*							$849,388	
M1.4.1		Percentage of direct implementation costs		5%	of	$16,987,765	$849,388		
M1.5	*Permitting*							$0	
M1.5.1.1		Permit Staffing	Permitting Manager	0	ls	$0	$0		Staffing costs included in tasks above
M1.5.1.2			Environmental Manager	0	ls	$0	$0		
M1.5.1.3			Technical Consultants	0	ls	$0	$0		
TOTAL								**$1,317,046**	

Worksheet 17 - Labour Rate Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

LABOUR RATES

Benefit Factors

Employment Insurance (EI) percentage:	1.63%	of base earnings	2017 Source: http://www.cra-arc.gc.ca/tx/bsnss/tpcs/pyrll/clcltng/ei/hstrc-eng.html
EI salary limit	$51,300		2017 Source: http://www.cra-arc.gc.ca/tx/bsnss/tpcs/pyrll/clcltng/ei/hstrc-eng.html
Canada Pension Plan (CPP) percentage:	4.95%	of base earnings	2017 Source: http://www.cra-arc.gc.ca/tx/bsnss/tpcs/pyrll/clcltng/cpp-rpc/hstrc-eng.html
CPP salary limit:	$55,300		2017 Source: http://www.cra-arc.gc.ca/tx/bsnss/tpcs/pyrll/clcltng/cpp-rpc/hstrc-eng.html
MSP/Health Benefits:	$0	per month	Yukon Territory has no employer or employee premiums or payroll taxes for health care (Source: CostMine 2014)
Small tools/safety gear allowance:	$250	per year	Estimated based on a BC mine site (CostMine 2014)
Workers Compensation:			2017 YWC rates http://www.wcb.yk.ca/QuestionResults/Assessments/Rates/Q0269.aspx
Resources (low)	3.17%		Metal mining, gravel crushing and screening, air services
Resources (med.)	4.79%		**DEFAULT** - Reclamation, Exploration, short haul trucking
Resources (high)	7.94%		Diamond drilling, forestry, long haul trucking, oil/gas
Services (med.)	1.42%		Catering, Housekeeping, Consultants doing fieldwork

							Rotation Details						Travel Costs						Overtime		Other		Hours				Annual Totals								
Cost Code	Item	Base Hourly Wage ($/hr)	Loading Rate (%)	Rate Used in Estimate ($/hr)	Source (See Notes)	Include Loading Calculations?	Rotation Type	Weeks Onsite per Year	Hours Worked per Day	Days per Week	Site Hours per Year	Travel Hours per year	Travel Time (1 way) (hrs)	Travel Origin	Flights (1 way)	Hotel (1 way)	Meals (1 way)	Total Per Rotation	Overtime Multiplier	Overtime Hourly Wage	Small Tool Allowance ?	Worker's Comp Classification	Base Work Hours per Year	Overtime Work Hours per Year	Vacation Hours	Stat Hours	Total Hours Paid per Year	Base Earnings ($/yr)	Base Salary	Overtime Salary	EI	CPP	Worker's Comp.	Small Tool Allowance	MSP/Health Benefits	Travel Cost	Annual Total
P.01	Camp Labourer	$23.72	150%	$35.53	1	Yes	2 in - 2 out	28	12	7	2,352	70	2.5	Pelly	$0	$0	$0	$0	1.5	$35.58	No	Services (med.)	1120	1470	80	88	2590	$55,789	$26,566	$52,303	$836	$2,737	$1,120	$0	$0	$0	$83,562
P.02	Design Engineer	$150.00	90%	$135.00	3, 4		2 in - 2 out	28	12	7	2,352	140	5	Vancouver	$400	$100	$50	$1,100	1	$150.00	No	Resources (med.)	1120	1620	160	88	2740	$352,800	$168,000	$243,000	$836	$2,737	$19,687	$0	$0	$154,000	$588,260
P.03	Driller / Blaster	$29.47	151%	$44.39	1	Yes	3 in - 2 out	35	12	7	2,940	93	4	Whitehorse	$0	$0	$0	$0	1.5	$44.21	Yes	Resources (med.)	1400	1801	80	88	3201	$86,642	$41,258	$79,628	$836	$2,737	$5,790	$250	$0	$0	$130,500
P.04	Engineering Technician (Consultant)	$130.00	90%	$117.00	3, 4																															
P.05	Environmental Monitor	$29.47	151%	$44.57	3	Yes	2 in - 2 out	28	12	7	2,352	112	4	Whitehorse	$0	$0	$0	$0	1.5	$44.21	No	Services (med.)	1120	1512	80	88	2632	$69,313	$33,006	$66,838	$836	$2,737	$1,418	$0	$0	$0	$104,836
P.06	Environmental Scientist (Consultant)	$130.00	90%	$117.00	3, 4																															
P.07	Foreman	$32.87	150%	$49.36	1	Yes	3 in - 2 out	35	12	7	2,940	93	4	Whitehorse	$0	$0	$0	$0	1.5	$49.31	Yes	Resources (med.)	1400	1801	80	88	3201	$96,638	$46,018	$88,815	$836	$2,737	$6,458	$250	$0	$0	$145,115
P.08	Head Camp Cook	$32.87	145%	$47.73	1	Yes	3 in - 2 out	35	12	7	2,940	93	4	Whitehorse	$0	$0	$0	$0	1.5	$49.31	No	Services (med.)	1400	1801	80	88	3201	$96,638	$46,018	$88,815	$836	$2,737	$1,915	$0	$0	$0	$140,321
P.09	Health and Safety Supervisor	$32.00	151%	$48.38	3	Yes	2 in - 2 out	28	12	7	2,352	112	4	Whitehorse	$0	$0	$0	$0	1.5	$48.00	Yes	Services (med.)	1120	1512	80	88	2632	$75,264	$35,840	$72,576	$836	$2,737	$1,540	$250	$0	$0	$113,779
P.10	Heavy Equip. Operator 1: Crane, Dragline, Shovels	$32.87	150%	$49.36	1	Yes	3 in - 2 out	35	12	7	2,940	93	4	Whitehorse	$0	$0	$0	$0	1.5	$49.31	Yes	Resources (med.)	1400	1801	80	88	3201	$96,638	$46,018	$88,815	$836	$2,737	$6,458	$250	$0	$0	$145,115
P.11	Heavy Equip. Operator 2: Excavator, Loader, Dozers	$29.47	151%	$44.39	1	Yes	3 in - 2 out	35	12	7	2,940	93	4	Whitehorse	$0	$0	$0	$0	1.5	$44.21	Yes	Resources (med.)	1400	1801	80	88	3201	$86,642	$41,258	$79,628	$836	$2,737	$5,790	$250	$0	$0	$130,500
P.12	Heavy Equip. Operator 3: Rollers and compactors	$23.72	150%	$35.54	1	Yes	3 in - 2 out	35	12	7	2,940	58	2.5	Pelly	$0	$0	$0	$0	1.5	$35.58	Yes	Resources (med.)	1400	1766	80	88	3166	$69,737	$33,208	$62,846	$836	$2,737	$4,601	$250	$0	$0	$104,479
P.13	Heavy Equip. Mechanic	$32.87	150%	$49.36	1	Yes	3 in - 2 out	35	12	7	2,940	93	4	Whitehorse	$0	$0	$0	$0	1.5	$49.31	Yes	Resources (med.)	1400	1801	80	88	3201	$96,638	$46,018	$88,815	$836	$2,737	$6,458	$250	$0	$0	$145,115
P.14	Heavy Equip. Servicer	$29.47	151%	$44.39	1	Yes	3 in - 2 out	35	12	7	2,940	93	4	Whitehorse	$0	$0	$0	$0	1.5	$44.21	Yes	Resources (med.)	1400	1801	80	88	3201	$86,642	$41,258	$79,628	$836	$2,737	$5,790	$250	$0	$0	$130,500
P.15	Helicopter Pilot	$32.87	148%	$48.53	3	Yes	3 in - 2 out	35	12	7	2,940	93	4	Whitehorse	$0	$0	$0	$0	1.5	$49.31	No	Resources (low)	1400	1801	80	88	3201	$96,638	$46,018	$88,815	$836	$2,737	$4,274	$0	$0	$0	$142,680
P.16	Labourers	$23.72	150%	$35.54	1	Yes	3 in - 2 out	35	12	7	2,940	58	2.5	Pelly	$0	$0	$0	$0	1.5	$35.58	Yes	Resources (med.)	1400	1766	80	88	3166	$69,737	$33,208	$62,846	$836	$2,737	$4,601	$250	$0	$0	$104,479
P.17	Linesperson (electric)	$32.87	150%	$49.36	1	Yes	3 in - 2 out	35	12	7	2,940	93	4	Whitehorse	$0	$0	$0	$0	1.5	$49.31	Yes	Resources (med.)	1400	1801	80	88	3201	$96,638	$46,018	$88,815	$836	$2,737	$6,458	$250	$0	$0	$145,115
P.18	Medic, First Aid Attendant	$23.72	157%	$37.32	1	Yes	2 in - 2 out	28	12	7	2,352	112	4	Whitehorse	$0	$0	$0	$0	1.5	$35.58	No	Resources (med.)	1120	1512	80	88	2632	$55,789	$26,566	$53,797	$836	$2,737	$3,849	$0	$0	$0	$87,786
P.19	Office Manager	$39.90	155%	$61.85	2	Yes	2 in - 2 out	28	12	7	2,352	112	4	Whitehorse	$0	$0	$0	$0	1.5	$59.85	Yes	Resources (med.)	1120	1512	80	88	2632	$93,845	$44,688	$90,493	$836	$2,737	$6,475	$250	$0	$0	$145,480
P.20	Owner's Project Manager	$39.90	155%	$61.85	2	Yes	2 in - 2 out	28	12	7	2,352	112	4	Whitehorse	$0	$0	$0	$0	1.5	$59.85	Yes	Resources (med.)	1120	1512	80	88	2632	$93,845	$44,688	$90,493	$836	$2,737	$6,475	$250	$0	$0	$145,480
P.21	Mine Manager	$42.00	155%	$65.02	2	Yes	2 in - 2 out	28	12	7	2,352	112	4	Whitehorse	$0	$0	$0	$0	1.5	$63.00	Yes	Resources (med.)	1120	1512	80	88	2632	$98,784	$47,040	$95,256	$836	$2,737	$6,816	$250	$0	$0	$152,936
P.22	Security Guard	$23.72	150%	$35.54	1	Yes	3 in - 2 out	35	12	7	2,940	58	2.5	Pelly	$0	$0	$0	$0	1.5	$35.58	Yes	Resources (med.)	1400	1766	80	88	3166	$69,737	$33,208	$62,846	$836	$2,737	$4,601	$250	$0	$0	$104,479
P.23	Site Caretaker	$26.14	151%	$39.52	3	Yes	3 in - 2 out	35	12	7	2,940	93	4	Whitehorse	$0	$0	$0	$0	1.5	$39.21	Yes	Resources (med.)	1400	1801	80	88	3201	$76,852	$36,596	$70,630	$836	$2,737	$5,136	$250	$0	$0	$116,186
P.24	Site Clerk / Administration	$29.47	156%	$46.11	3	Yes	2 in - 2 out	28	12	7	2,352	112	4	Whitehorse	$0	$0	$0	$0	1.5	$44.21	Yes	Resources (med.)	1120	1512	80	88	2632	$69,313	$33,006	$66,838	$836	$2,737	$4,783	$250	$0	$0	$108,450
P.25	Site Supervisor	$39.90	155%	$61.85	2	Yes	2 in - 2 out	28	12	7	2,352	112	4	Whitehorse	$0	$0	$0	$0	1.5	$59.85	Yes	Resources (med.)	1120	1512	80	88	2632	$93,845	$44,688	$90,493	$836	$2,737	$6,475	$250	$0	$0	$145,480
P.26	Surveyor	$32.87	150%	$49.36	1	Yes	3 in - 2 out	35	12	7	2,940	93	4	Whitehorse	$0	$0	$0	$0	1.5	$49.31	Yes	Resources (med.)	1400	1801	80	88	3201	$96,638	$46,018	$88,815	$836	$2,737	$6,458	$250	$0	$0	$145,115
P.27	Trades Labourer (carpenter, electrician, welder etc.)	$32.87	160%	$52.53	1	Yes	3 in - 2 out	35	12	7	2,940	93	4	Whitehorse	$0	$0	$50	$100	1.5	$49.31	Yes	Resources (med.)	1400	1801	80	88	3201	$96,638	$46,018	$88,815	$836	$2,737	$6,458	$250	$0	$9,333	$154,448
P.28	Truck Driver 1: 10 Ton GVW and up.	$29.47	149%	$43.84	1	Yes	3 in - 2 out	35	12	7	2,940	58	2.5	Pelly	$0	$0	$0	$0	1.5	$44.21	Yes	Resources (med.)	1400	1766	80	88	3166	$86,642	$41,258	$78,081	$836	$2,737	$5,716	$250	$0	$0	$128,879
P.29	Truck Driver 2: 3 - 10 Ton GVW	$26.14	149%	$39.03	1	Yes	3 in - 2 out	35	12	7	2,940	58	2.5	Pelly	$0	$0	$0	$0	1.5	$39.21	Yes	Resources (med.)	1400	1766	80	88	3166	$76,852	$36,596	$69,258	$836	$2,737	$5,070	$250	$0	$0	$114,748

Notes:
1 Basic Rate: Yukon Government Fair Wage Schedule - Effective Apr 1, 2016 (Accessed July 2016)
2 CostMine 2014 for an active BC mine site
3 Estimated from project experience
4 Consultants rates are reduced by 10% as this value is added back in later in the estimate (Contractor Profit = 10%)

Worksheet 18 - Equipment Rate Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

EQUIPMENT RATES

Adjustment Factors

Equipment Rates used in estimate: 3rd-Party - BC Blue Book
Fuel Consumption Rates: 2 - Medium — Note: Medium consumption applied to most situations
On-site fuel cost per litre (diesel): $0.96 — Average On-site Cost

BC Blue Book Cost Breakdown Assumptions

BC Blue Book Source: 2017-18 — 1.00
Owner overhead rate: 10% — BC Roadbuilder Association (2000)
Profit rate: 10% — BC Roadbuilder Association (2000)

BC Govt Fuel Price Adjustment Factors:
Avg. BC diesel cost: $1.03 — Average BC diesel cost for the past 12 months
On-Road Equipment ($/hr): $1.35
Off-road equipment ($/hr): $3.15
Date: 8/1/2017
Source: http://www2.gov.bc.ca/gov/content/industry/construction-industry/transportation-infrastructure/hired-equipment-program/fuel-price-adjustment

Equipment		Rate Summary				Operator Details			Weight	Fuel Consumption Details				BC Blue Book Cost Breakdown						
Equip Code	Model	Source Used in Estimate	Equipment Cost ($/hr)	Fuel Operating Cost ($/hr)	Total Equipment Rate ($/hr)	Number of Operators	Operator Type	Operator Rate ($/hr)	Shipping Weight (tonnes)	On-Road/ Off-Road	Low (liters/hr)	Medium (liters/hr)	High (liters/hr)	All-Found Rate ($/hr)	Operators Wage ($/hr)	Ownership/ Maintenance Cost ($/hr)	Fuel Operating Cost ($/hr)	Contractor Overhead Cost ($/hr)	Contractor Profit ($/hr)	Comments
	Backhoe																			
E.01.1	CAT430E (4x4)	3rd Party	$11.58	$11.66	$23.24	1	E. Operator 2	$44.39	7.4	Off	8.1	12.15	21.95	$91.65	$48.50	$11.58	$15.66	$7.57	$8.33	
	Compactors																			
E.02.1	CAT CP563 (Class 7 - 12-13.99 tons)	3rd Party	$29.58	$14.88	$44.46	1	Roller Operator	$35.54	11.5	Off	13	15.5	18.5	$117.60	$48.50	$29.58	$19.12	$9.72	$10.69	
E.02.2	Walk-behind vibrating (30 in)	3rd Party	$9.46	$2.25	$11.71	0	Roller Operator	$0.00	n/a	n/a	1.2	2.3	3.5	$11.45	$0.00	$9.46	$0.00	$0.95	$1.04	Fuel rate estimated from a 22 in Wacker Neuson (3.5L/hr under 100%operation)
	Dozers																			
E.03.1	CAT D6R	3rd Party	$105.36	$23.28	$128.64	1	E. Operator 2	$44.39	18.0	Off	18	24.25	30.5	$220.20	$48.50	$105.36	$28.13	$18.20	$20.02	
E.03.2	CAT D8R	3rd Party	$128.31	$35.28	$163.59	1	E. Operator 2	$44.39	37.6	Off	27.25	36.75	46.25	$263.55	$48.50	$128.31	$41.00	$21.78	$23.96	
E.03.3	CAT D10T	3rd Party	$194.68	$67.01	$261.69	1	E. Operator 2	$44.39	66.4	Off	51.45	69.8	88.6	$385.05	$48.50	$194.68	$75.04	$31.82	$35.00	
E.03.4	CAT D11R	3rd Party	$206.55	$96.00	$302.55	1	E. Operator 2	$44.39	104.6	Off	71.7	100	126.25	$437.05	$48.50	$206.55	$106.15	$36.12	$39.73	
	Drills																			
E.04.1	Air Rotary, 200 cfm compressor	3rd Party	$161.91	$24.00	$185.91	2	Driller / Blaster	$88.78	n/a	Off	20	25	30	$348.25	$97.00	$161.91	$28.90	$28.78	$31.66	Med. Fuel consumption rate estimated based on DR10 drill (http://www.orbitdrilling.com.au/mineralexp.html)
E.04.2	Air track production drill rig (900cfm)	3rd Party	$62.77	$48.00	$110.77	2	Driller / Blaster	$88.78	41.0	Off	33.35	50	66.5	$259.45	$97.00	$62.77	$54.65	$21.44	$23.59	Med. Fuel consumption rate estimated from Cost Mine 2015
	Excavators																			
E.05.1	CAT 330	3rd Party	$83.65	$29.76	$113.41	1	E. Operator 2	$44.39	35.1	Off	21.5	31	36.5	$202.35	$48.50	$83.65	$35.08	$16.72	$18.40	
E.05.2	CAT 385	3rd Party	$177.68	$62.40	$240.08	1	E. Operator 2	$44.39	83.0	Off	45.5	65	74.5	$358.50	$48.50	$177.68	$70.10	$29.63	$32.59	
	Graders																			
E.06.1	CAT 140M (Class 6 - 200-248 FWHP)	3rd Party	$78.57	$15.60	$94.17	1	E. Operator 2	$44.39	13.6	Off	10.35	16.25	24.2	$177.82	$48.50	$78.57	$19.89	$14.70	$16.17	10% grader attachment allowance added.
E.06.2	CAT 16M (Class 7 - 250-300 FWHP)	3rd Party	$83.88	$21.79	$105.67	1	E. Operator 2	$44.39	24.7	Off	14.7	22.7	37.25	$192.28	$48.50	$83.88	$26.53	$15.89	$17.48	10% grader attachment allowance added.
	Haul Trucks																			
E.07.1	Std Tandem Haul Truck (12 yd3)	3rd Party	$13.20	$11.52	$24.72	1	Truck Driver 1	$43.84	n/a	On	8.04	12	15.96	$91.25	$48.50	$13.20	$13.71	$7.54	$8.30	Fuel Consumption source: http://www.equipmentworld.com/owning-and-operating-costs-8/ (3.2 gal/hr for short hauls
E.07.2	CAT 735	3rd Party	$89.43	$25.87	$115.31	1	Truck Driver 1	$43.84	29.9	Off	16.55	26.95	37.35	$204.30	$48.50	$89.43	$30.91	$16.88	$18.57	
E.07.3	CAT 769D	3rd Party	$78.11	$34.27	$112.38	1	Truck Driver 1	$43.84	35.4	Off	25.5	35.7	45.9	$201.50	$48.50	$78.11	$39.92	$16.65	$18.32	
E.07.4	CAT 773G	3rd Party	$142.14	$48.77	$190.91	1	Truck Driver 1	$43.84	45.0	Off	36.25	50.8	65.35	$297.80	$48.50	$142.14	$55.47	$24.61	$27.07	
E.07.5	CAT 777D	3rd Party	$156.39	$63.02	$219.42	1	Truck Driver 1	$43.84	72.6	Off	46.85	65.65	84.4	$333.55	$48.50	$156.39	$70.77	$27.57	$30.32	
	Lifting																			
E.08.1	Hydraulic Crane, 30Ton	3rd Party	$81.38	$6.00	$87.38	1	E. Operator 1	$49.36	30.0	Off	4.19	6.25	8.31	$168.75	$48.50	$81.38	$9.59	$13.95	$15.34	Med. Fuel consumption rate estimated from CostMine 2015
	Loaders																			
E.09.1	CAT 950H	3rd Party	$61.19	$12.53	$73.72	1	E. Operator 2	$44.39	18.3	Off	9.65	13.05	16.6	$152.80	$48.50	$61.19	$16.59	$12.63	$13.89	
E.09.2	CAT 990	3rd Party	$104.53	$62.64	$167.17	1	E. Operator 2	$44.39	77.8	Off	50	65.25	84.75	$270.30	$48.50	$104.53	$70.36	$22.34	$24.57	
	Support Vehicles																			
E.10.1	Concrete Transit Mix Truck	3rd Party	$29.05	$10.85	$39.90	1	Labourer	$35.54	n/a	On	7.54	11.3	15.03	$109.55	$48.50	$29.05	$12.99	$9.05	$9.96	Med. Fuel consumption from NRCA (http://www.nrmca.org/operations/documents/2012fleetsurveyfinal repot.pdf)
E.10.2	Fuel Truck	3rd Party	$3.86	$9.25	$13.11	1	Equip. Servicer	$44.39	n/a	On	6.43	9.64	12.82	$77.00	$48.50	$3.86	$11.28	$6.36	$7.00	Med. Fuel consumption from CostMine2015 (mobile field fuel/lube truck)
E.10.3	Highway, Line Truck, with Aerial Platform	3rd Party	$58.08	$12.00	$70.08	1	Trade Labour	$52.53	n/a	On	8.34	12.5	16.63	$146.18	$48.50	$58.08	$14.23	$12.08	$13.29	Med. Fuel consumption from CostMine2015 (Flatbed truck)
E.10.4	Highway Service/Flatbed Truck (HIAB) 5 ton	3rd Party	$20.79	$12.00	$32.79	1	Equip. Servicer	$44.39	n/a	On	8.34	12.5	16.63	$101.05	$48.50	$20.79	$14.23	$8.35	$9.19	Med. Fuel consumption from CostMine2015 (Flatbed truck)
E.10.5	Lowbed Tractor/Trailer - 5 Axle (25 tonnes)	3rd Party	$28.59	$11.52	$40.11	1	Truck Driver 1	$43.84	n/a	Off	8.00	12	15.96	$112.05	$48.50	$28.59	$15.51	$9.26	$10.19	Fuel consumption estimated
E.10.6	Lowbed Tractor Trailer - 6 Axle (33 tonnes)	3rd Party	$34.64	$14.40	$49.04	1	Truck Driver 1	$43.84	n/a	Off	10.01	15	19.95	$123.10	$48.50	$34.64	$18.60	$10.17	$11.19	Fuel consumption estimated
E.10.7	Lowbed Tractor Trailer - 7 Axle (41 tonnes)	3rd Party	$49.22	$14.40	$63.62	1	Truck Driver 1	$43.84	n/a	Off	10.01	15	19.95	$140.75	$48.50	$49.22	$18.60	$11.63	$12.80	Fuel consumption estimated
E.10.8	Lowbed Tractor Trailer - 8 Axle (48 tonnes)	3rd Party	$51.78	$14.40	$66.18	1	Truck Driver 1	$43.84	n/a	Off	10.01	15	19.95	$143.85	$48.50	$51.78	$18.60	$11.89	$13.08	Fuel consumption estimated
E.10.9	Pick-up Truck (1.5T) 4x4	3rd Party	$7.55	$4.80	$12.35	0		$0.00	n/a	On	3	5	10	$17.00	$0.00	$7.55	$6.50	$1.40	$1.55	Fuel consumption rate estimated
E.10.10	Pilot Vehicle	3rd Party	$8.36	$4.80	$13.16	1	Labourer	$35.54	n/a	On	3	5	10	$55.50	$31.01	$8.36	$6.50	$4.59	$5.05	Fuel consumption rate estimated; BCBlueBook operator rate reduced to obtain a bare rate to match BlueBook rate
E.10.12	Water Truck (5,000 gal)	3rd Party	$15.49	$21.70	$37.18	1	Truck Driver 1	$43.84	n/a	Off	15.07	22.6	30.06	$109.40	$48.50	$15.49	$26.43	$9.04	$9.95	Med. Fuel consumption from CostMine2015 (5,000gel water truck)

Notes:

1 Fuel consumption obtained from the Caterpillar Performance Handbook Version 42.
2 All-found BC Blue Book rate excludes the fuel adjustment factor
3 Equipment weights from Caterpillar Performance Handbook Version 35; values of n/a means the equipment would be driven to site/towed by an pick-up.

	Misc. Equipment and Attachments	Unit	Equipment Rate ($/Unit)	Fuel Rate ($/Unit)	Total Equipment Rate ($/Unit)	Number of Operators	Operator Type	Operator Rate ($/hr)	Shipping Weight (tonnes)	Source
E.11.1	ATV (4 wheel)	month	$1,483.80	$20.00	$1,503.80	0		$0.00	0.3	BC Blue Book 2017-18; fuel rate estimated
E.11.2	Boomlift: 60' articulating	hr	$25.00	$2.00	$27.00	0		$0.00	10.0	United rental 2015; fuel rate estimated
E.11.3	Soil auger (Truck mounted	hr	$27.00	$0.00	$27.00	0		$0.00	n/a	Estimated
E.11.4	Bus: Passenger Bus (24 person cap.)	month	$1,658.85	$600.00	$2,258.85	0		$0.00	n/a	BC Blue Book 2017-18; fuel rate estimated
E.11.5	Butt-fusion welding machine	hr	$38.20	$9.55	$47.75	0		$0.00	n/a	RSMeans2016; fuel rate estimated
E.11.6	Cleaning Equipment - High pressure washer	hr	$15.75	$2.00	$17.75	0		$0.00	n/a	BC Blue Book 2017-18; fuel rate estimated, 2,000 gallons, includes 3 wands
E.11.7	Crusher (200 Tons/hr)	hr	$170.00	$35.00	$205.00	1	E. Operator 2	$44.39	25.0	RSMeans 2005 (17 03 9902)
E.11.8	Demolition shears (excavator attach.)	hr	$150.00	$0.00	$150.00	0		$0.00	7.0	Estimated; S390 (fits CAT 365-385)
E.11.9	Demolition grapples (excavator attach.)	hr	$21.00	$0.00	$21.00	0		$0.00	7.0	Estimated; G330 (fits CAT 345-365
E.11.10	Emergency Transport Vehicle	Month	$4,509.10	$10.00	$4,519.10	0		$0.00	n/a	BC Blue Book 2017-18; fuel rate estimated
E.11.11	Helicopter	hr	$1,500.00	$100.00	$1,600.00	1	Trade Labour	$52.53	n/a	Estimated from past project (2012)
E.11.12	Hydraulic hammer (excavator attachment)	hr	$87.05	$0.00	$87.05	0		$0.00	4.0	BC Blue Book 2017-18; 2,500 kg energy class
E.11.13	Hydraulic plate tamper (excavator attach.)	hr	$13.45	$0.00	$13.45	0		$0.00	3.0	BC Blue Book 2017-18; 454-680kg
E.11.14	Hydroseed Truck	hr	$47.75	$16.25	$39.90	0		$0.00	n/a	BC Blue Book 2017-18; 3,000gal water truck with highpressure spray nozzle
E.11.15	Light towers	month	$1,405.85	$272.16	$1,678.01	0		$0.00	1.0	BC Blue Book 2017-18; telescopic trailor mounted, multi-vaor light (3kV); (fuel assumes used average 5 hrs per day)
E.11.16	Power mulcher, (20 ton/hr, 115 HP	hr	$25.00	$24.33	$49.33	0		$0.00	n/a	RSMeans 2016 (015433202860)
E.11.17	Pump: mud pump 6 in discharge	hr	$12.98	$2.00	$14.98	0		$0.00	n/a	BC Blue Book 2017-18; fuel rate estimated
E.11.18	Screener (finley 583) (~400tonnes/hr)	hr	$189.59	$40.00	$223.00	1	E. Operator 2	$44.39	75.0	Nova Scotia Road Builder Association, 2012, fuel rate estimated
E.11.19	Spreader (for reveg appliction)	hr	$8.90	$0.00	$8.90	0		$0.00	n/a	BC Blue Book 2017-18; based on "hopper sander" for truck attachments
E.11.20	Welding outfit (truck mounted) - 400 amps	hr	$27.10	$0.00	$27.10	0		$0.00	n/a	BC Blue Book 2017-18
E.11.21	XRF rental	month	$1,272.00	$0.00	$1,272.00	0		$0.00	n/a	Quote: Pine Environmental (2010)

Worksheet 19 - Material Rates and Indirect Cost Inputs

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

MATERIAL RATES

RS Means Adjustment Factors

RSMeans Source: RSMeans Online (2016 QQ)

Notes:
1 All RSMeans 2017 material costs are in CAD and have been adjusted for location (Whitehorse).
2 As per note 1, a factor of 1.371 was applied to obtain a Yukon rate (compared to the US national average).
3 1.02 Inflation factor to convert from 2016 to 2018 dollars (obtained from Bank of Canada CPI calculator)

Cost Code	Item	Unit Cost	Unit	Additional Details	Source
M.01	Bitumen Geomembrane (BGM) Liner	$18.21	m2		Quote: Coletanche 2014, plus 15% shipment to site
M.02	Concrete: elevated slabs	$386.22	m3	Includes forms, grade 60 rebar, concrete, pla	RSMeans2016 (33053401900)
M.03	Erosion control mats (jute mesh), 4' wide	$0.83	m2		RSMeans2016 (312515160020), add 15% for shipment to site
M.04	Fertilizer	$1.12	kg		Access (2014) Closure Plan
M.05	Fertilizer tablets (for tree-planting)	$0.08			Quote: Pickseed (Prince George), 2014; factor of 15% applied for shipment to site
M.06	Formwork: curb forms, wood, 12" high, on grade, 1 use	$3.54	m2	Unit is m2 contact area	RSMeans2016 (31113652000)
M.07	Geosynthetic Clay Liner	$10.56	m2		Quote: Nilex Vancouver 15% added for shipment to site
M.08	Geotextile	$2.16	m2		Minto Quote: Layfield (2014) FOB Edmonton, 15% added for shipment to site
M.09	HDPE liner (60mil)	$9.38	m2		Minto Quote: Layfield (2014) FOB Edmonton, 15% added for shipment to site
M.10	HDPE pipe: 100mm	$20.57	m	Butt fusion joints, 40'length, SDR 21	RSMeans2016 (33 11 13-350-100)
M.11	HDPE pipe: 150mm	$57.38	m	Butt fusion joints, 40'length, SDR 21	RSMeans2016 (33 11 13-350-200)
M.12	HDPE pipe: 300mm	$184.68	m	Butt fusion joints, 40'length, SDR 21	RSMeans2016 (33 11 13-350-500)
M.13	Intake Structure: precast concrete panel wall	$37,000	each		Quote : Precon (2017) FOB Edmonton. Adjusted 15% for shipment
M.14	Lab testing: Hydrocarbons	$102	each		Estimated
M.15	Lab testing: SAL metals + pH (BC CSR package)	$133	each		Quote: CARO 2014 for BC CSR package, $30 allowance added for shipping
M.16	Mulch (for hydroseeding)	$1.02	kg		Quote: Pickseed (Prince George), 2014; factor of 15% applied for shipment to site
M.17	Rebar (#6 bar)	$3.37	m		RSMeans2016 (032111600400) , (#6 bar = 1.5 lbs per foot)
M.18	Seed/Fertilizer Mix 1: Dry mix	$16.12	kg		Pickseed (2015) onsite cost
M.19	Seed/Fertilizer Mix 2: Wet area mix	$17.85	kg		Access (2014) Closure Plan
M.20	Signage: Open Pit Warning signs (incl. posts)	$199.26	each		Includes concrete, post and sign
M.21	Signage: Large warning signs at entries (incl. posts)	$609.74	each		Assumed to be 3 times open pit signs.
M.22	Silt Fencing (3ft high)	$0.92	m		RSMeans2016 (31251416100), 15% added for shipment to site
M.23	Steel Beam: I-Beam W6x9	$16.09	m		RSMeans2016
M.24	Steel decking (q-deck):	$45.35	m2		RSMeans2016 (53133506700)
M.25	Steel Pipe: 6" dia.	$115.16	m		RSMeans2016 (221316202200)
M.26	Tackifier (for hydroseeding)	$3.69	kg		Quote: Pickseed (Prince George), 2014; factor of 15% applied for shipment to site
M.27	Wetland planting	$1,159.23	ha		EPA, 1997. Costs for Wetland Creation and Restoration Projects in the Glaciated Northeast. Average cost adjusted up by 37% for location factor, 15% for shipping, 42% for inflation.
M.28	Tree seedlings	$0.39	stem		Quote from a site in central BC in 2011; updated to 2016 dollars.

INDIRECT COSTS

Cost Code	Category	Rate Used in Estimate	Unit		Source/Comments
I.01	Barge Operating Costs	$10,832	month		Based on Minto site costs ($36,800/month), assumes is required 20 hours per week (down from existing 70hrs per week)
I.01	Bonding	3.0%	%	of direct costs	Past Project (2014)
I.02	Camp Operation Costs	$80	$/day/person	includes catering and housekeeping	Estimated; Note: this rate is higher than the exisitng camp cost and is conservatively estimated to account for the smaller camp size during closure/post-closure.
I.03	Camp power and heating costs	$3,570	month		Typical 2015 monthly cost between May and Oct 2015, and halved under assumption that only 1/2 of camp is required.
I.04	Communications	$1,000	month		Estimated
I.05	Contract Administration	5%	%	Of Direct and other indirect costs	
I.06	Contractor Profit	10%	%	Of Direct and other indirect costs	
I.07	Contingency	12%	%		
I.09	Engineering, Design, and Construction Plans	5%	%		allowance
I.11	Flight - Whitehorse-Minto Return	$3,000	ea		
I.13	Helicopter trip Carmacks-Minto Return	$3,000	ea		Based on site costs, assumes 2 hrs helicoptor time.
I.15	Freight	0%	%	Included in material costs	
I.17	Insurance	1.5%	%	Of labour costs	
I.19	Laboratory/Material Testing	$1,000	month		Estimated
I.21	Misc. Admin Supplies	$500	month		Estimated
I.23	Office Supplies	$1,000	month		Estimated

Worksheet 20 - Task Unit Rate Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

Cost Code	Item	Unit	Total Unit Cost	Manhours per Unit (hrs/Unit)	Labor Cost Per Unit ($/Unit)	Equipment Cost Per Unit ($/Unit)	Fuel Cost Per Unit ($/Unit)	Material Cost Per Unit ($/Unit)	Productivity (unit/hr)	Productivity/Crew/Fleet Source (RSMeans line item)	Equipment Fleet (Including Operators): Excavators: CAT 330	CAT 385	CAT 450E (4x4)	Haul Trucks: Std Tandem Haul Truck (12	CAT 735	Loader: CAT 950H	Dozers: CAT D6R	CAT D8R	CAT D10T	Compactor: CAT CP563 (Class 7 - 12-	Walk-behind vibrating (30 in)	Grader: CAT 140M (Class 6 - 200-	CAT 16M (Class 7 - 250-	Support Vehicles/Equipment: Hydraulic Crane, 30Ton	Concrete Transit Mix	Light Tower, Line Truck with	Aerial Platform Service/Flatbed Truck(HIAB) 5	Trailer - 6 Axle (23 tonnes)	Pick-up Truck (4x4) Utility	Universal shears	Excavator grapples	Hydraulic hammer	Other: Equip Cost ($/hr)	Fuel Cost ($/hr)	Description	Labour: Environmental Monitor	Foreman	Heavy Equip. Mechanic	Heavy Equip. Service	Labourers	Trainee Labourer	Material Details: Cost Code	Item	Unit Rate	Unit	Multiplier (unit/unit)	Other Comments/Details
Demolition/Equipment Removal																																															
C.1.01	Concrete foundations: break in place	**m2**	**$14.24**	0.081	$2.39	$9.37	$2.48	$0.00	37	RSMeans (02 41 6.17-0400)	1.0	1.0																				1.0																Crew B13L
C.1.02	Decontamination: Metal cleaning treatment, pressure washing	**tonne**	**$3.82**	0.072	$2.55	$1.13	$0.14	$0.00	13.94	RSMeans (05011051621																								$15.75	$2.00	Cleaning Equipment - High pressure washer					1.0								Based on RSMeans productivity of 15,000ft2/day, assumes avg. thickness of 1cm
C.1.03	Decontamination: Concrete cleaning treatment, pressure washing	**m2**	**$0.46**	0.009	$0.31	$0.14	$0.02	$0.00	116.13	RSMeans (05011051621																								$15.75	$2.00	Cleaning Equipment - High pressure washer					1.0								
C.1.04	Decontamination: General pressure washing	**hrs**	**$53.29**	1.000	$35.54	$15.75	$2.00	$0.00	1.00	n/a																								$15.75	$2.00	Cleaning Equipment - High pressure washer					1.0								
C.1.05	Demolition: Structural, steel	**tonne**	**$242.28**	2.000	$69.36	$139.08	$33.84	$0.00	5.0	Assumes average 50ton/shift (see right)		2.0			1.0	1.0									1.0							1.0	1.0										1.0			2.0					based on an average productivity for a XS4000 demolition shear (http://www.terrafirmaequipment.com/images/stories/virtuemart/product/series-ds-&-xs-shears.pdf)
C.1.06	Demolition: Structural, concrete	**m3**	**$94.92**	0.538	$15.92	$62.48	$16.53	$0.00	5.58	Based on foundation demolition, assu	1.0	1.0																				1.0																Crew B13L
C.1.07	Demolition: selective misc. demolition (wood)	**m3**	**$6.55**	0.099	$3.74	$2.07	$0.74	$0.00	40.38	RSMeans (24119180500)	1.0																																		3.0					Crew B3. (loader sub'ed for an excavator)
C.1.08	Demolition: Small wooden buildings (non selective demo)	**m3**	**$5.05**	0.076	$2.88	$1.60	$0.57	$0.00	52.39	RSMeans (24116130700)	1.0																																		3.0					Crew B3. small wood buildings (loader sub'ed for an excavator)
C.1.09	Disconnect services	**hrs**	**$100.42**	3.000	$88.07	$7.55	$4.80	$0.00	1.00	n/a																						1.0														1.0	1.0		
C.1.10	Electrical demolition: remove conduits (incl. fittings/hangers)	**km**	**$3,831.94**	68.957	$2,639.96	$940.09	$251.90	$0.00	0.20	Based in RSMeans (337139130150_							1.0								0.4		0.2	1.0		2.0												1.0			4.0	4.0				Based on RSMeans for conduit installation. Assumes 3 wires, dismantling is 50% more productive than installation
C.1.11	Electrical demolition: remove powerpoles	**each**	**$530.46**	9.143	$423.67	$88.51	$18.29	$0.00	0.66	Based in RSMeans (337116335140)																1.0																1.0			2.0	2.0				Based on RSMeans for 60' pole installation. Assumes dismantling is 50% more productive than installation
C.1.12	Electrical demolition: remove transformers	**each**	**$1,296.04**	23.323	$892.89	$317.96	$85.20	$0.00	0.58	Based in RSMeans (337323201200)							1.0								0.4		0.2	1.0		2.0												1.0			4.0	4.0				Based on RSMeans for transformer installation. Assumes dismantling is 50% more productive than installation
C.1.13	Equipment removal: Small equip., dismantle, prep for transport	**hrs**	**$268.71**	4.000	$168.00	$86.19	$14.53	$0.00	1	n/a						1.0																		$25.00	$2.00	Boomlift					2.0	1.0							
C.1.14	Equipment removal: Small equip., dismantle, prep for transport	**hrs**	**$457.98**	6.000	$269.89	$167.56	$20.53	$0.00	1	n/a						1.0								1.0										$25.00	$2.00	Boomlift					2.0	2.0							
C.1.15	HDPE Liner: cut and fold	**m2**	**$0.70**	0.012	$0.43	$0.19	$0.07	$0.00	430	Assumes can be removed 20 times q	1.0																																		4.0								
C.1.16	HDPE pipe demolition - 0.75 to 4 inches (19-100 mm)	**m**	**$5.15**	0.112	$5.15	$0.00	$0.00	$0.00	27	RSMeans2005 (02 41 13.38-1600)																																		1.0			1.0	1.0				
C.1.17	HDPE pipe demolition - 6 to 8 inches (150-200 mm)	**m**	**$7.21**	0.157	$7.21	$0.00	$0.00	$0.00	19	RSMeans2005 (02 41 13.38-1700)																																		1.0			1.0	1.0				
C.1.18	HDPE pipe demolition - 10 to 18 inches (250-450 mm)	**m**	**$12.02**	0.262	$12.02	$0.00	$0.00	$0.00	11	RSMeans2005 (02 41 13.38-1800)																																		1.0			1.0	1.0				
C.1.19	HDPE pipe demolition - 20 to 36 inches (500-915 mm)	**m**	**$18.04**	0.394	$18.04	$0.00	$0.00	$0.00	8	RSMeans2005 (02 41 13.38-1900)																																		1.0			1.0	1.0				
C.1.20	Modular Buildings: dismantle and load for transport	**each**	**$1,748.04**	19.500	$850.98	$730.60	$166.46	$0.00	0.33	Assumes 1 every 3 hours	1.0					1.0								1.0					0.5														2.0	1.0							
C.1.21	Pipelines: flushing	**each**	**$203.78**	4.000	$203.78	$0.00	$0.00	$0.00	0.5	Assumes 2hrs to flush 1 line																																		1.0				1.0				No real basis to task, but cost is expected to be small and insignificant
C.1.22	Reagents: Load for transport off-site	**hrs**	**$282.05**	4.000	$159.30	$95.83	$26.93	$0.00	1.0	n/a						1.0													1.0														2.0						No real basis to task, but cost is expected to be small and insignificant
C.1.23	Reclaim pump systems: Dismantle and prep for transport	**each**	**$1,284.65**	26.400	$1,205.96	$49.89	$28.80	$0.00	0.1	Assumed 6hr duration																				0.3												1.0			1.0	1.0				RSMEANS description: Remove pump to 200HP
C.1.24	Tailing system pumps: Dismantle and prep for transport	**each**	**$642.33**	13.200	$602.98	$24.94	$14.40	$0.00	0.3	RSMeans2005 (16010639)																				0.3												1.0			1.0	1.0				RSMEANS description: Remove pump to 200HP
Earthworks																																															
C.2.01	Bedding: Produce and stockpile	**m3**	**$3.51**	0.020	$0.96	$2.11	$0.44	$0.00	148	Based on avg 400tonne/hr screener						2.0																		$189.59	$40.00	Screen Plant						1.0								Assumes avg. material density of 1,800 km/s, 50min/hr, and 20% wastage
C.2.02	Berm: Construct berm (locally sourced, no truck required)	**m3**	**$1.56**	0.009	$0.42	$0.89	$0.25	$0.00	212	Factored productivity of CAT 330	1.0						1.0																																Berm shaped by excavator, dozer used to push suitable material to excavator
C.2.03	Clearing land: tree removal by dozer	**m2**	**$0.79**	0.004	$0.17	$0.49	$0.13	$0.00	263	RSMeans2005(31 13 13.20-0500)								1.0																																
C.2.04	Compact soil with vibratory plate (Hand): 15 cm lifts	**Cm3**	**$6.62**	0.097	$2.58	$3.01	$1.03	$0.00	30.96	RSMeans2005 (17-03-0508)	1.0										1.0																							1.0							
C.2.05	Compact soil with roller on flat ground: 15 cm lifts, 4 passes	**m2**	**$0.08**	0.001	$0.04	$0.02	$0.01	$0.00	1210.54	RSMeans2016 (312323235040)										1.0																								0.5						RSMeans crew: B-10Y
C.2.06	Compact soil with roller on slopes: 15 cm lifts, 4 passes (one way)	**m2**	**$0.20**	0.003	$0.11	$0.06	$0.03	$0.00	484.22	Based on flat ground, productivity adjusted by factor of 0.4										1.0																								0.5						RSMeans crew: B-10Y
C.2.07	Compaction, placement of materials in landfill	**m3**	**$13.04**	0.043	$1.89	$8.30	$2.86	$0.00	23.47	RSMeans 241191880500									1.0																														
C.2.08	Earthwork repair/maintenance fleet (post-closure)	**hrs**	**$476.08**	3.000	$132.06	$262.52	$81.50	$0.00	1	n/a	1.0				2.0																																		
C.2.08	Earthwork repair/maintenance fleet (perpetual maintenance)	**hrs**	**$649.11**	4.000	$176.45	$367.88	$104.78	$0.00	1	n/a	1.0				2.0		1.0																																
C.2.09	Erosion Control Mats (Jute mesh), Supply and install	**m2**	**$1.64**	0.018	$0.66	$0.09	$0.05	$0.83	228	RSMeans2016(312514160020)																				1.0												3.0		M.03	Erosion control mats (jute me	$0.83	m2	1.0	
C.2.10	Mulch: tree chipping/mulch production	**m3**	**$11.18**	0.113	$4.50	$4.87	$1.82	$0.00	35.5	Based on 20ton/hr mulcher. See note	0.5				0.5	1.0																		$25.00	$24.33	Power Mulcher					2.0								Applied factors of .833 for job efficiency, and .75 for equipment availability/operator skill.
C.2.11	Regrading - Rough grading to provide drainage (small areas)	**m2**	**$2.08**	0.010	$0.44	$1.28	$0.35	$0.00	100	Estimate								1.0																														
C.2.12	Regrading - Rough grading to provide drainage - yards	**m2**	**$0.42**	0.002	$0.09	$0.26	$0.07	$0.00	500	Estimate								1.0																														
C.2.13	Regrading - Waste Rock Dumps (D10)	**hrs**	**$306.07**	1.000	$44.39	$194.68	$67.01	$0.00	**1**	n/a									1.0																														Check that the dozer model matches that used on resloping worksheet
C.2.14	Rip-Rap (rounded, low quality): Screen and Stockpile	**m3**	**$7.32**	0.000	$1.78	$4.24	$1.30	$0.00	50	Based on 75% productivity of CAT 33	1.0							1.0																														
C.2.15	Rip-Rap (angular, high quality): Screen and Stockpile	**m3**	**$10.45**	0.000	$2.54	$6.06	$1.86	$0.00	35	Based on 50% productivity of CAT 33	1.0							1.0																														
C.2.16	Scarification: roads, yards, etc.	**m2**	**$0.12**	0.000	$0.03	$0.07	$0.02	$0.00	1277	RSMeans2016(32 91 13-2620													1.0																												
C.2.17	Silt fencing: Supply, Install, and remove	**m**	**$6.75**	0.040	$3.23	$1.93	$0.68	$0.92	50	RSMeans2016 (312514131100)			1.0										1.0																							2.0		M.22	Silt Fencing (3ft high)	$0.92	m	1.0	
C.2.18	Surface preparation for liner	**m2**	**$1.31**	0.034	$1.31	$0.00	$0.00	$0.00	73.125	RSMeans2013 (31 22 16.10-1150)																																0.5			2.0					RSMeans = Fine/finish grading, hand grading for slab on grade
C.2.19	Underground plug placement with small dozer	**hrs**	**$173.02**	0.000	$44.39	$105.36	$23.28	$0.00	1	n/a							1.0																														
Environmental Monitoring/Investigations																																															
C.3.01	Contaminated soil investigation test pits	**ea**	**$293.25**	0.500	$22.24	$2.89	$2.92	$265.20	**4**	4 per hour			1.0																												1.0					M.15	Lab testing: SAL metals + ph	$132.60	each	2.0	
C.3.02	Contaminated soil confirmation testing	**ea**	**$22.01**	0.167	$7.43	$1.33	$0.00	$13.26	**6**	6 per hour																								$7.95		XRF rental					1.0					M.15	Lab testing: SAL metals + ph	$132.60	each	0.1	Assumes on-site testing using XRF, and 1 out of every 10 samples submitted for laboratory analysis
C.3.03	HC soil test pits	**ea**	**$232.05**	0.500	$22.24	$2.89	$2.92	$204.00	**4**	4 per hour			1.0																												1.0					M.14	Lab testing: Hydrocarbons	$102.00	each	2.0	
C.3.04	Aerate soils in landfarm	**event**	**$2,255.90**	22.500	$999.20	$923.10	$333.60	$0.00	**0.1**	Assumed 10 hours per event (incl. mo	1.0																			0.3											1.0								
C.3.05	Landfarm confirmation testing	**yr**	**$2,121.63**	16.000	$711.69	$92.63	$93.31	$1,224.00	**0.125**	Assumed 8 hours for sampling			1.0																												1.0					M.14	Lab testing: Hydrocarbons	$102.00	each	12.0	Costs exclude reporting
Groundwater / Piping																																															
C.4.01	HDPE pipe: 100mm; supplied and installed	**m**	**$40.12**	0.328	$14.26	$4.50	$0.79	$20.57	**15.2**	RSMeans2016 (33 11 13-350-100)																				1.0					$47.75		Butt fusion machine		1.0			2.0	1.0	M.10	HDPE pipe: 100mm	$20.57	m	1	Crew B-22A
C.4.02	HDPE pipe: 150mm; supplied and installed	**m**	**$77.95**	0.345	$15.01	$4.73	$0.83	$57.38	**14.5**	RSMeans2016 (33 11 13-350-200)																				1.0					$47.75		Butt fusion machine		1.0			2.0	1.0	M.11	HDPE pipe: 150mm	$57.38	m	1.0	Crew B-22A
C.4.05	HDPE pipe: 300mm, perforated	**m**	**$214.75**	0.505	$21.94	$6.92	$1.21	$184.68	**9.9**	RSMeans2016 (33 11 13-350-500)																				1.0					$47.75		Butt fusion machine		1.0			2.0	1.0	M.12	HDPE pipe: 300mm	$184.68	m	1.0	Crew B-22A
C.4.11	Pump water: small 2" centrifugal pump (100GPM) (small head)	**m3**	**$0.35**	0.002	$0.08	$0.23	$0.05	$0.00	22	100 gallons per minute																								$5.00	$1.00	Small 2" pump					0.1								
C.4.12	Pump water: large 6" mud pump	**m3**	**$0.07**	0.000	$0.01	$0.05	$0.01	$0.00	273	1200 gallons per minute																								$12.98	$2.00	6" mud Pump					0.1								
C.4.13	Pump water: operate large 6" mud pump	**hr**	**$18.53**	0.100	$3.55	$12.98	$2.00	$0.00	1	n/a																								$12.98	$2.00	6" mud Pump					0.1								
C.4.14	Pump water: Set-up pumps hoses, etc.	**hr**	**$147.76**	4.000	$120.43	$20.53	$6.80	$0.00	1	n/a																					1.0			$12.98	$2.00	6" mud Pump		1.0			2.0								

Cost Code	Item	Unit	Total Unit Cost	Manhours per Unit (hrs/Unit)	Labor Cost Per Unit ($/Unit)	Equipment Cost Per Unit ($/Unit)	Fuel Cost Per Unit ($/Unit)	Material Cost Per Unit ($/Unit)	Productivity (unit/hr)	Productivity/Crew/Fleet Source (RSMeans line item)	Equipment Fleet (Including Operators): Excavators: CAT 330	CAT 385	CAT430E (4x4)	Haul Trucks: Std Tandem Haul Truck (12	CAT 735	Loader: CAT 950H	Dozers: CAT D6R	CAT D8R	CAT D10T	Compactor: CAT CP563 (Class 7 - 12- Walk-behind vibrating (30 in)	Grader: CAT 140M (Class 7 - 200-	CAT 16M (Class 7 - 250-	Support Vehicles/Equipment: Hydraulic Crane, 30Ton	Concrete Transit Mix Truck	Tight Kevy Line Truck with	Aerial Platform Service/Flatbed Truck (HAB) 5	Trailer - 6 Axle (53 tonnes),	Pick-up Truck (4 x 4) Utility Truck	Other: Concrete shears (excavator	grapples (excavator	hammer (excavator	Equip Cost ($/hr)	Fuel Cost ($/hr)	Description	Labour: Environmental Monitor	Foreman	Heavy Equip. Mechanic	Heavy Equip. Servicer	Labourers	Trades Labourer (carpenter	Material Details: Cost Code	Item	Unit Rate	Unit	Multiplier (unit/unit)	Other Comments/Details
Materials																																															
C.5.01	BGM Liner: Supplied and Installed	m2	**$21.22**	0.047	$1.93	$0.79	$0.28	$18.21	105.5	SRK measured productivity (2015)	1.0																										3.0	1.0	M.01	Bitumen Geomembrane (BG	$18.21	m2	1.0	Productivity measured by SRK on a 2015 cover installation project
C.5.07	Geotextile: Supply and Install (large areas, no seaming)	m2	**$2.82**	0.014	$0.55	$0.05	$0.06	$2.16	211	Assumed 2x value below (assumes no seaming			1.0																								2.0		M.08	Geotextile	$2.16	m2	1.0	
C.5.08	Geotextile: Supply and Install (channels)	m2	**$3.47**	0.028	$1.09	$0.11	$0.11	$2.16	105.5	Productivity adjusted to match installed quote ($4			1.0																								2.0		M.08	Geotextile	$2.16	m2	1.0	
C.5.09	Geosynthetic Clay Liner: Supplied & Installed	m2	**$12.07**	0.033	$1.36	$0.08	$0.08	$10.56	150	Estimated from a 2014 quote for a cover installati			1.0																								3.0	1.0	M.07	Geosynthetic Clay Liner	$10.56	m2	1.0	
C.5.10	HDPE liner (60mil): Supplied & Installed (large areas)	m2	**$11.53**	0.047	$1.93	$0.11	$0.11	$9.38	105.5	Productivity adjusted to match BGM			1.0																								3.0	1.0	M.09	HDPE liner (60mil)	$9.38	m2	1.0	
C.5.11	HDPE liner: Supplied & Installed (channels)	m2	**$14.66**	0.116	$4.74	$0.27	$0.27	$9.38	43	RSMeans2016(33-47-13.53-1100)			1.0																								3.0	1.0	M.09	HDPE liner (60mil)	$9.38	m2	1.0	
C.5.12	Intake structure: Supply and Install (precast concrete)	each	**$47,172.29**	140.00	$5,946	$3,583	$644	$37,000	0.05	SRK measured productivity (2011) (2 days)			1.0			1.0						1.0												$25.00	$2.00	Boomlift					3.0	1.0	M.13	Intake Structure: precast cor	$37,000	each	1.0	Two days installation
C.5.13	Signage: Open Pit warning signs	each	**$261.00**	0.928	$40.27	$15.90	$5.57	$199.26	**4.3**	RSMeans2016(32-31-13.30-6600)																2.0								$27.00		Auger attachment		1.0			1.0		M.20	Signage: Open Pit Warning s	$199.26	each	1.0	8 foot galvanized steel post set in concrete with a 18"x24" reflective warning sign
C.5.14	Signage: Large signs at main entrance	each	**$671.48**	0.928	$40.27	$15.90	$5.57	$609.74	**4.3**	RSMeans2016(32-31-13.30-6600)																2.0								$27.00		Auger attachment		1.0			1.0		M.21	Signage: Large warning sign	$609.74	each	1.0	8 foot galvanized steel post set in concrete with a 18"x24" reflective warning sign
Revegetation																																														
C.6.01	Seeding/Fertilizing: Tractor Application, Dry seed mix	ha	**$2,669.89**	33.126	$1,029.50	$401.72	$232.94	$1,005.72	0.060375	RSMeans2016 (32 92 19-14-0100)			1.0																0.5				$8.90		Spreader					0.5		n/a	Seed Mix #1	$1,005.72	ha	1.0	See 'SeedMix' worksheet for seed mix details and application rates
C.6.02	Seeding/Fertilizing: Tractor Application, Wet seed mix	ha	**$2,669.89**	33.126	$1,029.50	$401.72	$232.94	$1,005.72	0.060375	RSMeans2016 (32 92 19-14-0100)			1.0																0.5				$8.90		Spreader					0.5		n/a	Seed Mix #2	$1,005.72	ha	1.0	See 'SeedMix' worksheet for seed mix details and application rates
C.6.03	Seeding/Fertilizing: Hydroseeding Application, Dry seed mix	ha	**$4,841.11**	32.292	$1,242.82	$653.23	$129.17	$2,815.89	0.0929	RSMeans2016 (32 92 19.14-0200)																1.0								$39.90		Water truck w/ nozzle					2.0		n/a	Seed Mix #3	$2,815.89	ha	1.0	See 'SeedMix' worksheet for seed mix details and application rates
C.6.04	Seeding/Fertilizing: Hydroseeding Application, Wet Area mix	ha	**$4,841.11**	32.292	$1,242.82	$653.23	$129.17	$2,815.89	0.0929	RSMeans2016 (32 92 19.14-0200)																1.0								$39.90		Water truck w/ nozzle					2.0		n/a	Seed Mix #4	$2,815.89	ha	1.0	See 'SeedMix' worksheet for seed mix details and application rates
C.6.05	Wetland planting: by hand	ha	**$3,089.53**	43.750	$1,443.27	$307.03	$180.00	$1,159.23	**0.08**	Assume 50% productivity as tree planting																1.0		0.5													2.0		M.27	Wetland planting	$1,159.23	ha	1.0	
C.6.06	Tree planting (1,000 stems/ha)	ha	**$2,343.03**	51.250	$1,832.33	$25.98	$15.00	$469.71	**0.16**	8 planters, 8hrs/day. WorkSafeBC: avg. tree planter plants 1,600 stems/day																0.2													8.0		n/a	Tree planting #1	$469.71	ha	1.0	See 'SeedMix' worksheet for seed mix details and application rates A pickup truck and 0.2 time for support flatbed truck added to move materials around site
Transportation																																														
C.7.01	Misc. On-site Mobilization/Demobilization	hr	**$250.67**	2.000	$88.22	$118.29	$44.16	$0.00	1.0000	n/a	1.0																	1.0																			
C.7.02	Off-site disposal to Whitehorse in 6-Axle Semi-Trailer (by weight)	tonne	**$38.41**	0.400	$17.56	$15.18	$5.67	$0.00	2.8571	Assumes 80% capacity/7 hr travel/offload time						0.1												1.0																			
C.7.03	Off-site Debris disposal to Whitehorse in 6-Axle Semi-Trailer (by volume)	m3	**$25.67**	0.267	$11.74	$10.15	$3.79	$0.00	4.2745	Assumes 80% capacity/7 hr travel/offload time						0.1												1.0																			
C.7.04	Off-site disposal to Whitehorse in 6-Axle Semi-Trailer (by trip)	trip	**$768.21**	8.000	$351.24	$303.64	$113.33	$0.00	0.1429	Assumes 7 hr travel/offload time						0.1												1.0																			
C.7.05	Off-site disposal to Whitehorse in 8-Axle Semi-Trailer (by weight)	tonne	**$23.23**	0.208	$9.25	$11.03	$2.95	$0.00	5.4857	Assumes 80% capacity/7 hr travel/offload time						0.1																				$51.78	$14.40	8 axle lowboy trailer			1.0										
C.7.06	Off-site Debris disposal to Whitehorse in 8-Axle Semi-Trailer (by volume)	m3	**$15.53**	0.139	$6.18	$7.37	$1.97	$0.00	8.2070	Assumes 80% capacity/7 hr travel/offload time						0.1																				$51.78	$14.40	8 axle lowboy trailer			1.0										
C.7.07	Off-site disposal to Whitehorse in 8-Axle Semi-Trailer (by trip)	trip	**$892.11**	8.000	$355.10	$423.68	$113.33	$0.00	0.1429	Assumes 7 hr travel/offload time						0.1																				$51.78	$14.40	8 axle lowboy trailer			1.0										
Underground Opening Seals																																														
C.8.01	Excavate to competent soils around perimeter and clean (pressure wash)	m3	**$104.49**	2.000	$84.00	$13.66	$6.83	$0.00	2	Assumed			1.0																						$15.75	$2.00	quipment - High pressure washer					2.0	1.0						
C.8.02	Structural I-Beams: supply and install	m	**$88.27**	1.225	$50.19	$15.95	$6.05	$16.09	5.7	RSMeans 51223750100 (see notes)	1.0																		1.0								1.0			1.0	3.0	M.23	Steel Beam: I-Beam W6x9	$16.09	m	1.0	Productivity was reduced by 75% to account for dangerous conditions, and small job
C.8.03	Formwork: concrete ringwall	m2	**$312.97**	6.408	$309.43	$0.00	$0.00	$3.54	0.6	RSMeans 31113652000 (see notes)																														1.0	3.0	M.06	Formwork: curb forms, wooc	$3.54	m2	1.0	Productivity was reduced by 75% to account for dangerous conditions, and small job
C.8.04	Steel decking (q-deck): supply and install	m2	**$87.87**	0.574	$29.08	$10.60	$2.85	$45.35	10.5	RSMeans 53133506700	1.0																								$27.10	$0.00	utlit (truck mounted) - 400 an	1.0				4.0		M.24	Steel decking (q-deck):	$45.35	m2	1.0	Productivity was reduced by 75% to account for dangerous conditions, and small job
C.8.05	Rebar: supplied and installed	m	**$4.64**	0.024	$1.27	$0.00	$0.00	$3.37	165	RSMeans 32111600400 (see notes)																															4.0	M.17	Rebar (#6 bar)	$3.37	m	1.0	#6 rebar assumed (1.5 lbs per ft.)
C.8.06	Concrete: Supply and install elevated slab (incl. formwork, rebar)	m3	**$729.58**	7.076	$329.23	$10.63	$3.50	$386.22	3.6744	RSMeans 33053401900													1.0												$10.00	$2.00	Gas Eng. Vibrator		1.0			8.0	16.0	M.02	Concrete: elevated slabs	$386.22	m3	1.0	
C.8.07	Vent raise: supply and install	ea	**$478.39**	0.500	$17.77	$0.00	$0.00	$460.62	2	assumed																															1.0	M.25	Steel Pipe: 6" dia.	$115.16	m	4.0	

Worksheet 21 - Relocation Unit Rates

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date: February 6, 2018
File Location: \\van-svr0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate

						Unit Rate Summary								Fleet					
Cost Code	Source	Destination	Activities	Material	Costed Volume Unit	Total Unit Rate ($/m^3)	Manhours (hrs/m^3)	Labor Rate ($/m^3)	Equipment Rate ($/m^3)	Fuel Rate ($/m^3)	Productivity (m^3/hr)	Total Dist. (1-way) (km)	Avg. Grade (%)	Excavator	Truck Fleet	Max. # of Trucks	# of Dozers	Dozer Size	Compaction?
R.001	Misc. 1km flat haul w/ Articulated Trucks		Load, haul, dump, spread	Earth, moist	CCM	$5.31	0.03	$1.39	$3.03	$0.90	159	1.0	0.0%	CAT 330	CAT 735	5	1	CAT D8R	No
R.002	Misc. 1km flat haul w/ Articulated Trucks w/ compaction		Load, haul, dump, spread, compact	Earth, moist	CCM	$5.82	0.04	$1.61	$3.21	$0.99	159	1.0	0.0%	CAT 330	CAT 735	5	1	CAT D8R	Yes
R.003	Misc. 1km flat haul w/ Large Fleet		Load, haul, dump, spread	Earth, moist	CCM	$2.78	0.01	$0.51	$1.70	$0.57	516	1.0	0.0%	CAT 385	CAT 773G	5	1	CAT D8R	No
R.004	Misc. 1km flat haul w/ Large Fleet		Load, haul, dump, spread	Earth, moist	CCM	$2.78	0.01	$0.51	$1.70	$0.57	516	1.0	0.0%	CAT 385	CAT 773G	5	1	CAT D8R	No
R.005	Airport Laydown	Solid Waste Landfill	Load, haul, dump	Demo Debris	LCM	$6.37	0.04	$1.76	$3.53	$1.08	100	2.4	3.2%	CAT 330	CAT 735	5	0	CAT D8R	No
R.006	Airstrip Area	Solid Waste Landfill	Load, haul, dump	Demo Debris	LCM	$5.35	0.03	$1.48	$2.95	$0.92	89	1.0	0.0%	CAT 330	CAT 735	5	0	CAT D8R	No
R.007	Camp Area	Solid Waste Landfill	Load, haul, dump	Demo Debris	LCM	$7.20	0.05	$1.99	$3.99	$1.22	88	3.1	2.7%	CAT 330	CAT 735	5	0	CAT D8R	No
R.008	Dyno Compound	Solid Waste Landfill	Load, haul, dump	Demo Debris	LCM	$8.39	0.05	$2.32	$4.65	$1.42	76	5.5	-0.5%	CAT 330	CAT 735	5	0	CAT D8R	No
R.009	Fuel Farm Area	Solid Waste Landfill	Load, haul, dump	Demo Debris	LCM	$8.37	0.05	$2.32	$4.64	$1.41	76	3.6	2.3%	CAT 330	CAT 735	5	0	CAT D8R	No
R.010	Fuel Farm Area	Landfarm	Load, haul, dump	Earth, moist	CCM	$7.84	0.05	$2.17	$4.36	$1.31	122	3.6	2.3%	CAT 330	CAT 735	5	0	CAT D8R	No
R.011	IROD Laydown	SWD-High Grade Waste	Load, haul, dump, spread, compact	Sand, dry	CCM	$3.14	0.01	$0.63	$1.89	$0.63	547	1.6	-0.8%	CAT 385	CAT 773G	5	2	CAT D8R	Yes
R.012	Main Pit (SAT)	Main Pit (SAT)	Load, haul, dump	Waste Rock (Minto	CCM	$3.13	0.02	$0.75	$1.86	$0.51	409	0.4	0.0%	CAT 385	CAT 735	3	3	CAT D6R	No
R.013	Main Pit Dump	Main Pit Dump Buttress	Load, haul, dump	Waste Rock (Minto	CCM	$3.08	0.01	$0.58	$1.89	$0.61	456	1.0	-5.0%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.014	Main Pit Dump	Main Pit Tailings Cover	Load, haul, dump, spread	Waste Rock (Minto	CCM	$5.05	0.03	$1.38	$2.86	$0.82	160	1.5	-5.0%	CAT 330	CAT 735	5	1	CAT D6R	No
R.015	Main Waste Dump Extension	Main Pit Access Point	Load, haul, dump, place	Rip-rap (Minto)	CCM	$4.77	0.03	$1.25	$2.71	$0.81	177	1.8	-6.0%	CAT 330	CAT 735	5	1	CAT D8R	No
R.016	Main Waste Dump Extension	Area 2 Pit Access point	Load, haul, dump, place	Rip-rap (Minto)	CCM	$4.57	0.03	$1.19	$2.60	$0.77	184	1.6	-6.0%	CAT 330	CAT 735	5	1	CAT D8R	No
R.017	Main Waste Dump Extension	Area 118 Pit Access Point	Load, haul, dump, place	Rip-rap (Minto)	CCM	$6.45	0.04	$1.70	$3.66	$1.08	156	3.2	-1.4%	CAT 330	CAT 735	5	1	CAT D8R	No
R.018	Main Waste Dump Extension	W15/Main Pit Channel	Load, haul, dump	Rip-rap (Minto)	CCM	$4.47	0.03	$1.17	$2.55	$0.76	189	1.2	-10.0%	CAT 330	CAT 735	5	1	CAT D8R	No
R.019	Main Waste Dump Extension	W15/Main Pit Channel	Load, haul, dump, place	Sand & gravel, dry	CCM	$6.51	0.04	$1.64	$3.77	$1.10	162	1.2	-10.0%	CAT 330	CAT 735	5	2	CAT D8R	No
R.020	Main Waste Dump Extension	Ditch B1 (W35 to Mill area)	Load, haul, dump	Rip-rap (Minto)	CCM	$5.03	0.03	$1.32	$2.87	$0.85	168	2.0	-4.8%	CAT 330	CAT 735	5	1	CAT D8R	No
R.021	Main Waste Dump Extension	Ditch B1 (W35 to Mill area)	Load, haul, dump, place	Sand & gravel, dry	CCM	$7.41	0.04	$1.86	$4.29	$1.25	142	2.0	-4.8%	CAT 330	CAT 735	5	2	CAT D8R	No
R.022	Main Waste Dump Extension	Mill Area	Load, haul, dump	Rip-rap (Minto)	CCM	$5.85	0.04	$1.54	$3.33	$0.98	171	2.1	-5.3%	CAT 330	CAT 735	5	1	CAT D8R	No
R.023	Main Waste Dump Extension	Mill Area	Load, haul, dump, place	Sand & gravel, dry	CCM	$8.50	0.05	$2.17	$4.90	$1.43	142	2.1	-5.3%	CAT 330	CAT 735	5	2	CAT D8R	No
R.024	Main Waste Dump Extension	Minto North	Load, haul, dump	Rip-rap (Minto)	CCM	$4.14	0.02	$1.08	$2.36	$0.70	204	1.1	-5.0%	CAT 330	CAT 735	5	1	CAT D8R	No
R.025	Main Waste Dump Extension	MVFE2 (by access road)	Load, haul, dump	Rip-rap (Minto)	CCM	$6.87	0.04	$1.81	$3.91	$1.16	146	2.5	-4.9%	CAT 330	CAT 735	5	1	CAT D8R	No
R.026	Main Waste Dump Extension	MVFE2 (by access road)	Load, haul, dump, place	Sand & gravel, dry	CCM	$9.11	0.05	$2.34	$5.24	$1.53	150	2.5	-4.9%	CAT 330	CAT 735	5	2	CAT D8R	No
R.027	Main Waste Dump Extension	MVFE2 Toe area	Load, haul, dump	Rip-rap (Minto)	CCM	$7.00	0.04	$1.86	$3.97	$1.17	166	2.9	-5.0%	CAT 330	CAT 735	5	1	CAT D8R	No
R.028	Main Waste Dump Extension	MVFE2 Toe area	Load, haul, dump, place	Sand & gravel, dry	CCM	$10.05	0.06	$2.59	$5.78	$1.69	136	2.9	-5.0%	CAT 330	CAT 735	5	2	CAT D8R	No
R.029	Main Waste Dump Extension	Main Waste Dump	Load, haul, dump	Rip-rap (Minto)	CCM	$4.24	0.02	$1.09	$2.42	$0.72	161	0.4	-6.0%	CAT 330	CAT 735	5	1	CAT D8R	No
R.030	Main Waste Dump Extension	Main Waste Dump	Load, haul, dump	Sand & gravel, dry	CCM	$4.83	0.03	$1.25	$2.76	$0.82	142	0.4	-6.0%	CAT 330	CAT 735	5	1	CAT D8R	No
R.031	Main Waste Dump Extension	Southwest Dump	Load, haul, dump	Rip-rap (Minto)	CCM	$5.54	0.03	$1.45	$3.15	$0.94	152	2.0	-3.5%	CAT 330	CAT 735	5	1	CAT D8R	No
R.032	Main Waste Dump Extension	Southwest Dump	Load, haul, dump	Sand & gravel, dry	CCM	$5.87	0.04	$1.55	$3.34	$0.99	171	2.0	-3.5%	CAT 330	CAT 735	5	1	CAT D8R	No
R.033	Main Waste Dump Extension	Reclamation OVB Dump	Load, haul, dump	Rip-rap (Minto)	CCM	$5.30	0.03	$1.39	$3.02	$0.90	159	1.3	1.0%	CAT 330	CAT 735	5	1	CAT D8R	No
R.034	Main Waste Dump Extension	Reclamation OVB Dump	Load, haul, dump, place	Sand & gravel, dry	CCM	$6.29	0.04	$1.64	$3.58	$1.06	134	1.3	1.0%	CAT 330	CAT 735	5	1	CAT D8R	No
R.035	Main Waste Dump Extension	Tailings Diversion Ditch	Load, haul, dump	Rip-rap (Minto)	CCM	$7.79	0.05	$2.07	$4.42	$1.30	149	3.3	-4.8%	CAT 330	CAT 735	5	1	CAT D8R	No
R.036	MWD Low Grade Stockpile	Main Pit	Load, haul, dump, spread	Waste Rock (Minto	CCM	$3.02	0.01	$0.56	$1.84	$0.62	474	1.9	-5.4%	CAT 385	CAT 773G	5	1	CAT D8R	No
R.037	Mill Area	Solid Waste Landfill	Load, haul, dump	Demo Debris	LCM	$6.77	0.04	$1.87	$3.75	$1.14	94	3.0	3.1%	CAT 330	CAT 735	5	0	CAT D8R	No
R.038	Mill Area	Landfarm	Load, haul, dump	Waste Rock (Minto	CCM	$5.70	0.04	$1.58	$3.17	$0.96	139	3.0	3.1%	CAT 330	CAT 735	5	0	CAT D8R	No
R.039	Mill Area	Minto South Portal	Load, haul, dump	Waste Rock (Minto	CCM	$5.84	0.03	$1.53	$3.33	$0.99	144	1.6	3.8%	CAT 330	CAT 735	5	1	CAT D8R	No
R.040	Mill Pond	Main Pit	Load, haul, dump	Sand, wet	CCM	$3.30	0.02	$0.91	$1.82	$0.56	144	0.4	5.0%	CAT 330	CAT 735	5	0	CAT D8R	No
R.041	Minto North	MVFE Stage 2	Load, haul, dump, spread	Waste Rock (Minto	CCM	$5.15	0.02	$0.97	$3.15	$1.04	364	4.1	-3.1%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.042	Minto North Portal	Solid Waste Landfill	Load, haul, dump	Demo Debris	LCM	$8.18	0.05	$2.27	$4.53	$1.38	78	5.7	0.2%	CAT 330	CAT 735	5	0	CAT D8R	No
R.043	Minto South Portal	Solid Waste Landfill	Load, haul, dump	Demo Debris	LCM	$7.18	0.05	$1.99	$3.98	$1.21	88	3.2	0.3%	CAT 330	CAT 735	5	0	CAT D8R	No
R.044	Ore Stockpile Area (Mill)	Main Pit	Load, haul, dump, spread	Waste Rock (Minto	CCM	$2.82	0.01	$0.52	$1.72	$0.57	425	1.1	1.5%	CAT 385	CAT 773G	5	1	CAT D8R	No
R.045	Pelly Laydown	Solid Waste Landfill	Load, haul, dump	Demo Debris	LCM	$7.33	0.05	$2.03	$4.06	$1.24	87	4.3	-0.7%	CAT 330	CAT 735	5	0	CAT D8R	No
R.046	Reclamation Ovb. Dump	Camp Area	Load, haul, dump, spread	Earth, moist	CCM	$4.50	0.02	$0.85	$2.75	$0.91	416	2.9	-4.3%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.047	Reclamation Ovb. Dump	Camp Area	Load, haul, dump	Earth, moist	CCM	$4.00	0.02	$0.74	$2.44	$0.82	416	2.9	-4.3%	CAT 385	CAT 773G	5	1	CAT D8R	No
R.048	Reclamation Ovb. Dump	DSTSF	Load, haul, dump, spread	Earth, moist	CCM	$4.55	0.02	$0.86	$2.78	$0.91	412	3.1	-4.2%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.049	Reclamation Ovb. Dump	Fuel Storage Area	Load, haul, dump, spread	Earth, moist	CCM	$4.56	0.02	$0.86	$2.79	$0.92	411	2.7	-3.7%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.050	Reclamation Ovb. Dump	Main Dump	Load, haul, dump, spread	Earth, moist	CCM	$4.37	0.02	$0.82	$2.67	$0.88	429	1.5	2.0%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.051	Reclamation Ovb. Dump	Main Dump	Load, haul, dump	Earth, moist	CCM	$3.89	0.02	$0.72	$2.37	$0.80	429	1.5	2.0%	CAT 385	CAT 773G	5	1	CAT D8R	No
R.052	Reclamation Ovb. Dump	Main Pit	Load, haul, dump, spread	Earth, moist	CCM	$4.18	0.02	$0.79	$2.55	$0.84	448	2.6	-4.8%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.053	Reclamation Ovb. Dump	Main Pit Dump	Load, haul, dump, spread	Earth, moist	CCM	$3.79	0.02	$0.71	$2.32	$0.76	433	1.7	-5.5%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.054	Reclamation Ovb. Dump	Main Pit Dump	Load, haul, dump	Earth, moist	CCM	$3.31	0.01	$0.61	$2.02	$0.68	433	1.7	-5.5%	CAT 385	CAT 773G	5	1	CAT D8R	No
R.055	Reclamation Ovb. Dump	Mill Area	Load, haul, dump, spread	Earth, moist	CCM	$4.25	0.02	$0.80	$2.60	$0.86	441	2.7	-4.8%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.056	Reclamation Ovb. Dump	MVFE Stage 1 and 2	Load, haul, dump, spread	Earth, moist	CCM	$4.77	0.02	$0.90	$2.91	$0.96	393	3.4	-3.8%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.057	Reclamation Ovb. Dump	MVFE Stage 1 and 2	Load, haul, dump	Earth, moist	CCM	$4.24	0.02	$0.78	$2.59	$0.87	393	3.4	-3.8%	CAT 385	CAT 773G	5	1	CAT D8R	No
R.058	Reclamation Ovb. Dump	Pelly Laydown	Load, haul, dump, spread	Earth, moist	CCM	$3.45	0.01	$0.65	$2.11	$0.69	475	1.4	-4.5%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.059	Reclamation Ovb. Dump	Southwest Dump	Load, haul, dump, spread	Earth, moist	CCM	$4.08	0.02	$0.77	$2.49	$0.82	402	1.7	0.2%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.060	Reclamation Ovb. Dump	SWD-High Grade Waste	Load, haul, dump, spread	Earth, moist	CCM	$3.87	0.02	$0.73	$2.37	$0.78	423	1.9	-0.8%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.061	Reclamation Ovb. Dump	SWD-High Grade Waste	Load, haul, dump	Earth, moist	CCM	$3.30	0.01	$0.61	$2.02	$0.68	433	2.0	-1.0%	CAT 385	CAT 773G	5	1	CAT D8R	No

R.062	Reclamation Ovb. Dump	Ridgetop Pit	Load, haul, dump, spread	Earth, moist	CCM	$6.61	0.03	$1.24	$4.04	$1.33	283	4.6	1.5%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.063	Reclamation Ovb. Dump	Ridgetop Waste Dump	Load, haul, dump, spread	Earth, moist	CCM	$6.27	0.03	$1.18	$3.83	$1.26	299	4.0	1.7%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.064	Reclamation Ovb. Dump	Ridgetop Waste Dump	Load, haul, dump	Earth, moist	CCM	$5.57	0.02	$1.03	$3.40	$1.14	299	4.0	1.7%	CAT 385	CAT 773G	5	1	CAT D8R	No
R.065	Reclamation Ovb. Dump	W15 Sump Area	Load, haul, dump, spread	Earth, moist	CCM	$3.56	0.02	$0.67	$2.18	$0.71	460	1.5	-4.8%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.066	Reclamation Ovb. Dump	Tailings Diversion Ditch	Load, haul, dump, spread	Earth, moist	CCM	$5.23	0.02	$0.98	$3.20	$1.05	358	4.1	-3.2%	CAT 385	CAT 773G	5	2	CAT D8R	No
R.067	Reclamation Ovb. Dump	Tailings Diversion Ditch	Load, haul, dump	Earth, moist	CCM	$4.65	0.02	$0.86	$2.84	$0.95	358	4.1	-3.2%	CAT 385	CAT 773G	5	1	CAT D8R	No
R.068	Ridgetop Waste Dump	Ridgetop Pit	Load, haul, dump	Waste Rock (Minto	CCM	$3.12	0.01	$0.58	$1.90	$0.64	459	2.0	-4.4%	CAT 385	CAT 773G	5	1	CAT D8R	No
R.069	SWD-High Grade Waste	Main Pit	Load, haul, dump, spread	Waste Rock (Minto	CCM	$3.59	0.02	$0.66	$2.19	$0.73	399	2.8	-3.7%	CAT 385	CAT 773G	5	1	CAT D8R	No
R.070	W15 Sump	Solid Waste Landfill	Load, haul, dump	Demo Debris	LCM	$6.86	0.04	$1.90	$3.80	$1.16	93	3.7	1.1%	CAT 330	CAT 735	5	0	CAT D8R	No
	Note: Leave this row blank - Insert new rows above this one																		

EXCAVATION UNIT RATES (NO HAULING)

				Unit Rate Summary						Inputs				
Cost Code	Description	Material	Costed Volume Unit	Total Unit Rate ($/m^3)	Manhours (hrs/m^3)	Labor Rate ($/m^3)	Equipment Rate ($/m^3)	Fuel Rate ($/m^3)	Productivity (m^3/hr)	Excavator	# of Dozers	Dozer Size	Job Condition	Source, Comments/Notes
R.901	Culvert removal, large excavator	Waste Rock (Minto	BCM	$0.93	0.00	$0.15	$0.58	$0.20	305	CAT 385	0	CAT D8R	Avg. (Typical	
R.902	Channel excavation, small excavator, dozer used to spoil locally	Earth, moist	BCM	$2.88	0.02	$0.77	$1.64	$0.46	115	CAT 330	1	CAT D6R	Below Avg.	Below average conditions used to account for channel shaping
R.903	Channel excavation in steep areas, small excavator, dozer used to spoil locally	Earth, moist	BCM	$3.26	0.02	$0.88	$1.86	$0.52	101	CAT 330	1	CAT D6R	Severe	Severe conditions used to account for channel shaping, difficult access
R.904	Channel rip-rap placement	Rip-rap (Minto)	CCM	$1.72	0.01	$0.48	$0.91	$0.32	92	CAT 330	0	CAT D6R	Below Avg.	Below average conditions used to account for channel shaping/placement
R.905	Channel rip-rap placement in difficult access areas	Rip-rap (Minto)	CCM	$4.08	0.02	$1.10	$2.33	$0.65	81	CAT 330	1	CAT D6R	Severe	Severe conditions used to account for channel shaping/placement, dozer added to help
Note: Leave this row blank - Insert new rows above this one.														

Notes:

Worksheet 22 - Area Calculations for Covers and Revegetation

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

References (dwgs/plans) \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\!040_AutoCAD\1CM002-45_ClosureQuantities.dwg

A: Existing Yards, Waste Rock Dumps

	Areas						Reveg Prescription						Calculated Quantities							
Facility	Total 2D Area (pre-regrade) (m2)	2D Slope Area (from 'Regrade' worksheet) (m^2)	3D Slope Area (from 'Regrade' worksheet) (m^2)	Footprint Area increase due to regrading (from 'Regrade' worksheet) (m^2)	3D slope Area override (m^2)	Flat Area (post-regrade) override (m^2)	Flat Area Cover Thickness (m)	Slope Area Cover Thickness (m)	Flat Area Scarified/ Ripped?	Slopes Scarified/ Ripped?	Percentage of flat areas seeded/fertilized	Percentage of sloped Areas seeded/fertilized?	Estimated Flat Area (post-regrade)	3D Sloped Area (m^2)	Scarified flat areas (m^2)	Scarified sloped areas (m^2)	Cover Volume in Flat Areas (m^3)	Cover Volume - Sloped Areas (m^3)	Revegetated Area (ha)	Comments/Notes
Airport Laydown	21,277								Yes		100%		21,277	0	21,277	0	0	0	2.13	
Airstrip	64,378								Yes		100%		64,378	0	64,378	0	0	0	6.44	
Area 118 Backfill Dump	41,899			0			0	0	Yes	Yes	100%	100%	41,899	0	41,899	0	0	0	4.19	
Camp	33,525	17,739	18,699	4,840			0.5	0.5	Yes	Yes	100%	100%	20,626	18,699	20,626	18,699	10,313	9,349	3.93	
Crusher Area	26,964			0	0	26,964	0.5		Yes	Yes	100%	100%	26,964	0	26,964	0	13,482	0	2.70	
DSTSF WR Shell	50,237	51,376	52,957			0	0.5	0.5	Yes	No	100%	100%	0	52,957	0	0	0	26,479	5.30	
DSTSF Tailings Surface	191,561	10,000	10,308			48,000	0.5	0.5	No	No	100%	100%	48,000	10,308	0	0	24,000	5,154	19.16	Assumes areas covered in residuum only require covers. Note: Reveg area formula has been adjusted compared to the other rows so that the entire DSTSF area is included in the reveg area calculation.
Explosive Plant & Storage Areas	55,113						0	0	Yes	No	100%	100%	55,113	0	55,113	0	0	0	5.51	Roads between laydown/storage areas included in 'Roads' worksheet
Fuel Storage Area	8,400						0.5	0	No	No	100%	0%	8,400	0	0	0	4,200	0	0.84	
Ice Rich Overburden Dump	27,583				0		0	0	No	No	100%	100%	27,583	0	0	0	0	0	2.76	
IROD Laydown	23,055	4,949	5,147	0			0	0	Yes	No	100%	100%	18,106	5,147	18,106	0	0	0	2.33	Roads between laydown/storage areas included in 'Roads' worksheet
Main Waste Dump Ext. (includes Main Dump)	496,414	280,341	295,506	9,053			0.5	0.5	No	No	100%	100%	225,126	295,506	0	0	112,563	147,753	52.06	
Main Waste Dump Wrap	98,046	81,600	84,895	0			0.5	0.5	No	No	100%	100%	16,446	84,895	0	0	8,223	42,447	10.13	
Main Pit Dump (inc. rev.5-6)	236,535	159,480	168,106	13,485			0.5	0.5	No	No	100%	100%	90,540	168,106	0	0	45,270	84,053	25.86	
Mill Area	42,365						0.5	0.5	Yes	Yes	100%	100%	42,365	0	42,365	0	21,182	0	4.24	
Mill Valley Fill Extension (1 and 2)	171,634				59,852	114,853	0.5	0.5	No	No	100%	100%	114,853	59,852	0	0	57,427	29,926	17.47	
Minto South Portal	105,446	45,406	47,862	2,725			0	0	Yes	No	100%	100%	62,765	47,862	62,765	0	0	0	11.06	
Ore Stockpile Area	139,307				0		0.5		Yes	No	100%	100%	139,307	0	139,307	0	69,653	0	13.93	
Pelly Laydown	51,710						0.5	0.5	Yes	No	100%	100%	51,710	0	51,710	0	25,855	0	5.17	
Reclamation OVB Dump	297,857				0	297,857			No	No	100%	100%	297,857	0	0	0	0	0	29.79	
Ridgetop Exploration Area	34,778				0				Yes	No	100%	100%	34,778	0	34,778	0	0	0	3.48	Assumes 50% of area scarified/reveged.
Ridgetop Waste Dump	286,836	271,319	285,996	28,609			0.5	0.5	No	No	100%	100%	44,126	285,996	0	0	22,063	142,998	33.01	
Ridgetop South Backfill Dump	30,662	28,172	29,696	6,001			0	0	No	No	100%	100%	8,491	29,696	0	0	0	0	3.82	
Southwest Dump	627,621				0		0.5	0.5	No	No	100%	100%	627,621	0	0	0	313,810	0	62.76	Excludes High Grade Waste Stockpile
SWD - High Grade Waste (In-place)	31,165	40,679	41,485	15,430			1	1	No	No	100%	100%	5,915	41,485	0	0	5,915	41,485	4.74	
SWD - High Grade Waste (removed)	31,165						0.5	0.5	No	No	100%	100%	31,165	0	0	0	15,583	0	3.12	
Tailings Diversion Ditch	72,143	34,246	36,099	12,467			0.5	0.5	No	No	100%	100%	50,364	36,099	0	0	25,182	18,049	8.65	
WSP Dam Breach	13,950				13,950	0	0.5	0.5	No	No	100%	100%	0	13,950	0	0	0	6,975	1.40	Cover volume is an allowance: it is unlikely it will be to be placed over a large portion of the dam footpri due to slopes steeper than 2H:1V
W15 Sump Area	12,306	4,235	4,465	1,413			0.5	0.5	Yes	No	100%	100%	9,484	4,465	9,484	0	4,742	2,232	1.39	
TOTAL																	**779,463**	**556,901**	**347**	

Notes:

B: Future Yards, Waste Rock Dumps

	Areas						Reveg Prescription						Calculated Quantities							
Facility	Total 2D Area (pre-regrade) (m^2)	2D Slope Area (from 'Regrade' worksheet) (m^2)	3D Slope Area (from 'Regrade' worksheet) (m^2)	Footprint Area increase due to regrading (from 'Regrade' worksheet) (m^2)	3D slope Area override (m^2)	Flat Area (post-regrade) override (m^2)	Flat Area Cover Thickness (m)	Slope Area Cover Thickness (m)	Flat Area Scarified/ Ripped?	Slopes Scarified/ Ripped?	Percentage of flat areas seeded/fertilized	Percentage of sloped Areas seeded/fertilized?	Estimated Flat Area (post-regrade)	3D Sloped Area (m^2)	Scarified flat areas (m^2)	Scarified sloped areas (m^2)	Cover Volume Flat Areas (m^2)	Cover Volume - Sloped Areas (m^3)	Revegetated Area (ha)	Comments/Notes
Area 118 Backfill Dump	68,438	18,405	19,401	0			0	0	Yes	Yes	100%	100%	50,033	19,401	50,033	19,401	0	0	6.94	

C: Complex Covers

	Inputs							Calculated Quantities											
Facility	Flat Area (m^2)	3D Slope Area (from 'Regrade' worksheet) (m^2)	Bedding Layer Thickness (m)	Liner (Y/N)	Geotextile Layers	Protection Layer Thickness (m)	Growth Medium Cover Thickness (m)	Liner Area (m^2)	Geotextile Area (m^2)	Bedding Volume - Flat Areas (m^3)	Bedding Volume - Sloped Areas (m^3)	Total Bedding Volume (m^3)	Protection Layer Volume - Flat Areas (m^3)	Protection Layer Volume - Sloped Areas (m^3)	Total Protection Layer Volume (m^3)	Growth Medium Volume - Flat Areas (m^3)	Growth Medium Volume - Sloped Areas (m^3)	Total Growth Medium Volume (m^3)	Revegetated Area (ha)	Comments/Notes
Main Pit Tailings	17,750			No	0	1	0.5	0	0	0	0	0	17,750	0	17,750	8,875	0	8,875	1.78	Protection layer = waste rock
Ridgetop Tailings	62,477			No	0	1	0.5	0	0	0	0	0	62,477	0	62,477	31,239	0	31,239	6.25	

Worksheet 23 - Demolition Quantity Calculations

Project: Minto Mine Closure Cost Estimate
Project No.: 1CM002.P04
Client: Minto Explorations Ltd.
Date of Submission: August 2016
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

TOTAL DEMOLITION/SOLID WASTE VOLUME FOR SOLID WASTE LANDFILL 12,920 LCM

A: Building Demolition

				Exterior Building Dimensions					Sub-component Measurements							Disposal	Steel Quantity Calcs			Concrete Quantity Calcs		Wood/Misc. Qnty Calcs.		Off-site Transport		On-site Disposal	
No.	Area	Sub Area	Sub-component/ Building/ Section	Building Demolition Type	Length (m)	Width (m)	Height (m)	Modular Building Loads	Material	Wall thickness (m)[1]	# of Exterior Steel Columns	Avg. Ext. Column X-Section Area (m2)	# of Steel Trusses	Average Truss x section area (m2)	Disposal Location	Wall mass (tonnes)	Column and Truss Mass (tonnes)	Total Mass (tonnes)	In-Place Concrete Volume (m3)	Concrete Debris Volume (LCM)	In-Place Demo Volume (m3)	Debris Volume (LCM)	Modular Building Truckloads (Offsite)	Off-site Debris Trips	On-site Demolition Debris Volume (LCM)	Comments/Notes
Airport Laydown																										
1	Airport Laydown	Mechanics Shop	Entire building	Steel Building - Small	24.00	25.00	6.00		Steel	0.01					On-site	95.6	0.0	95.6	0.0	0.0	0.0	0.0	0.0	0.0	137	AutoCAD/Photos
2		Covered Storage		Wooden Building - Large	70.00	4.00	2.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	560.0	184.8	0.0	0.0	185	Photos
SUBTOTAL - Airport Laydown																		**96**		**0**	**560**		**0**	**0**	**321**	
Airstrip Area																										
1	Airstrip	Exploration Camp	Camp tents (11)	Wooden Building - Small	44.00	4.00	2.50		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	440.0	145.2	0.0	0.0	145	AutoCAD/Photos (lengths summed to account for 11 structures
2			Exploration core shed 1	Wooden Building - Small	7.50	4.50	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	101.3	33.4	0.0	0.0	33	AutoCAD/Photos
3			Exploration core shed 2	Wooden Building - Small	10.00	4.50	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	135.0	44.6	0.0	0.0	45	AutoCAD/Photos
4			Core Logging Shack	Wooden Building - Small	25.00	6.30	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	472.5	155.9	0.0	0.0	156	AutoCAD/Photos
SUBTOTAL - Airstrip																		**0**		**0**	**1149**		**0**	**0**	**379**	
Camp Area																										
1	Offices	Safety Building	Prefab Bldg. Shell	Steel Building - Small	17.00	18.00	6.00		Steel	0.03					On-site	175.3	0.0	175.3	0.0	0.0	0.0	0.0	0.0	0.0	250	AutoCAD/Photos
2		Mine Offices	Entire building	Wooden Building - Small	50.00	10.00	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	1500.0	495.0	0.0	0.0	495	AutoCAD/Photos
3	Old Camp Complex	Capstone Building	Entire building	Wooden Building - Small	56.00	9.00	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	1512.0	499.0	0.0	0.0	499	AutoCAD/Photos
4		Building betw Capstone & Sherwood		Wooden Building - Small	19.00	7.00	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	399.0	131.7	0.0	0.0	132	AutoCAD/Photos
5		Sherwood Building		Wooden Building - Small	110.00	7.50	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	2475.0	816.8	0.0	0.0	817	AutoCAD/Photos
6		Minto Building		Wooden Building - Small	70.00	7.50	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	1575.0	519.8	0.0	0.0	520	AutoCAD/Photos
7		Dining Hall/Dry		Wooden Building - Small	64.00	9.00	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	1728.0	570.2	0.0	0.0	570	AutoCAD/Photos
8		Gym/Muster		Wooden Building - Small	16.00	8.00	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	384.0	126.7	0.0	0.0	127	AutoCAD/Photos
9		Site Services/IT Offices		Wooden Building - Small	28.00	9.00	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	756.0	249.5	0.0	0.0	249	AutoCAD/Photos
10	North End of Camp	Trailer betw/Minto and Selkirk Tower		Wooden Building - Small	30.00	7.00	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	630.0	207.9	0.0	0.0	208	AutoCAD/Photos
11		Selkirk Towers		Modular Buildings - Dismantle/Prep				87							Off-site								87.0			Photos
12	Other	Misc. Seacans		Modular Buildings - Dismantle/Prep				5							Off-site								5.0			Photos
SUBTOTAL - Camp Area																		**175**		**0**	**10959**		**92**	**0**	**3867**	
Explosive Plant and Storage Areas																										
1	Explosive Plant	Dyno Compound	Dyno Garage	Steel Building - Small	15.00	15.00	6.00		Steel	0.01					On-site	47.1	0.0	47.1	0.0	0.0	0.0	0.0	0.0	0.0	67	AutoCAD/Photos
2		Dyno office		Modular Buildings - Dismantle/Prep				1							Off-site								1.0			Photos
SUBTOTAL - Explosive Plant and Storage Areas																		**47**		**0**	**0**		**1**	**0**	**67**	
Fuel Farm Area																										
1	Fuel Farm Area	Generators	Sheds	Wooden Building - Small	5.00	5.00	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	75.0	24.8	0.0	0.0	25	AutoCAD/Photos
SUBTOTAL - Fuel Farm Area																		**0**		**0**	**75**		**0**	**0**	**25**	
Mill Area Complex																										
1	Mill Complex	Process Building	Prefab Bldg. Shell	Steel Building - Large	59.76	21.34	15.24		Steel	0.03	20	0.0055914	10	0.0223656	On-site	905.1	52.1	957.2	0.0	0.0	0.0	0.0	0.0	0.0	1367	Drawings: 16-10-001 and 105
2		Concentrator Storage	Bayline 1	Steel Building - Large	30.49	6.71	9.15		Steel	0.03					On-site	213.7	0.0	213.7	0.0	0.0	0.0	0.0	0.0	0.0	305	Drawing 34-14-004
3		Tailings filter building	Prefab Bldg. Shell	Steel Building - Large	36.5854	20	15.24		Steel	0.03					On-site	593.3	0.0	593.3	0.0	0.0	0.0	0.0	0.0	0.0	848	Drawings 35-10-106 and 35-10-104
4		Mill Building	Phase 2 shell	Steel Building - Large	24.3902	14.4817	13.72		Steel	0.03	5	0.0055914	10	0.0223656	On-site	342.9	29.2	372.0	0.0	0.0	0.0	0.0	0.0	0.0	531	Drawings: 16-10-001 and 105
5		Mill Warehouse	Shell	Steel Building - Large	30.4878	24.3902	7.32		Steel	0.03					On-site	373.5	0.0	373.5	0.0	0.0	0.0	0.0	0.0	0.0	534	
6		Assay Lab	Entire building	Wooden Building - Small	17	10	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	510.0	168.3	0.0	0.0	168	
7		Mechanics Shop	Shell	Steel Building - Small	14	20	5.00		Steel	0.03					On-site	149.7	0.0	149.7	0.0	0.0	0.0	0.0	0.0	0.0	214	AutoCAD/Photos
8		Electricians shop	Entire building	Wooden Building - Small	15	10	3.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	450.0	148.5	0.0	0.0	149	AutoCAD/Photos
SUBTOTAL - Mill Area Complex																		**2660**		**0**	**960**		**0**	**0**	**4116**	
Mill Valley Fill Laydown																										
1	Reagent Storage Area		Tent	Wooden Building - Small	25.00	18.00	4.00		Wood/Mixed	0.01					On-site	0.0	0.0	0.0	0.0	0.0	1800.0	594.0	0.0	0.0	594	height negatively adjusted to reduce demo vol.
SUBTOTAL - Mill Valley Fill Laydown																		**0**		**0**	**1800**		**0**	**0**	**594**	
Pelly Laydown																										
1	Pelly Laydown	Pelly office	3 modular trailers	Modular Buildings - Dismantle/Prep				3							Off-site								3.0			AutoCAD/Photos
2		Pelly workshop	Wooden building	Wooden Building - Large	12.00	21.00	6.00		Wood/Mixed Debris						On-site	0.0	0.0	0.0	0.0	0.0	1512.0	499.0	0.0	0.0	499	AutoCAD/Photos
3			prefab building shell	Steel Building - Small	15.00	30.00	6.00		Steel	0.01					On-site	79.7	0.0	79.7	0.0	0.0	0.0	0.0	0.0	0.0	114	AutoCAD/Photos
4		Pelly warehouse	Weatherhaven	Wooden Building - Small	40.00	25.00	4.00		Wood/Mixed	0.03					On-site	0.0	0.0	0.0	0.0	0.0	4000.0	1320.0	0.0	0.0	1320	AutoCAD/Photos
5	W15 Sump Area	W15 Laydown	4 propane tanks	Modular Buildings - Dismantle/Prep				4							Off-site								4.0			AutoCAD/Photos
SUBTOTAL - Pelly Laydown																		**80**		**0**	**5512**		**7**	**0**	**1933**	
Water Treatment Plant																										
1	Water Treatment Plant		Prefab Bldg. Shell	Steel Building - Large	46.04	18.29	9.15		Steel	0.01					Off-site	162.5	0.0	162.5	0.0	0.0	0.0	0.0	0.0	3.4	0	
SUBTOTAL - Water Treatment Plant																		**163**		**0**	**0**		**0**	**3**	**0**	

Notes:
1. Internal steel structure/demo debris accounted for in part C
2. Steel wall thickness adjusted to account for airspace, insulation, HVAC debris, etc.
3. Wall areas includes the roof
4. Building debris volume for wooden structures based on FEMA "Debris Estimating Field Guide" FEMA publication No. 329.

B: Concrete Foundations

				Disposal	Exterior Building		Foundation Dimensions					Calculations					
No.	Area	Sub Area	Sub-component/ Building/ Section	Disposal Location	Length (m)	Width (m)	Slab Thickness (m)	Wall Thickness (m)	Wall Height (m)	Requires washing/de contaminati on?	In-place Concrete Vol. User Override (m3)	In-Place Concrete Volume (m3)	Concrete Debris Volume (m3)	Total Mass (tonnes)	Washing Area (m2)	On-site Demolition Debris Volume (LCM)	Comments/Notes
Camp Area																	
1	Offices	Safety Building	Concrete foundation	Bury-in-place	17	18	0.15	0.25	0.3	No		51.2	86.1	122.8	0.0	0.0	
SUBTOTAL - Camp Area												**51**	**86**	**123**	**0**	**0**	
Mill Area Complex																	
1	Mill Complex	Crusher and Ore Storage	Concrete - Ore Dump	Bury-in-place	12	6	0.3	0.305	1	No		32.6	54.9	78.2	0.0	0.0	
2		Mill Building	phase 1 portion	Bury-in-place	59.7561	21.3415	0.203	0.254	0.3	Yes		271.6	457.4	651.8	1275.3	0.0	Dwg. 16-10-022 and 16--10-001
3		Mill Building	Phase 2 portion	Bury-in-place	24.3902	14.4817	0.203	0.254	0.3	Yes		77.6	130.7	186.3	353.2	0.0	Dwg. 16-10-022 and 16--10-001
4		Mill Building	Compressor/generator area	Bury-in-place	12.1951	10.061	0.203	0.254	0.3	Yes		28.3	47.7	67.9	122.7	0.0	Dwg. 16-10-022 and 16--10-001
5		Concentrate Storage	Concrete foundation	Bury-in-place	73.7805	30.4878	0.3	0.45	2.21037	Yes		882.2	1485.9	2117.4	2249.4	0.0	Dwg. 34-15-001
6		Tailings Filter Building	Concrete foundation	Bury-in-place	36.5854	20	0.3	0.305	0.3	Yes		229.9	387.1	551.7	731.7	0.0	Dwg 35-10-022 and 35-10-101
7		Mill Warehouse	Concrete foundation	Bury-in-place	30.4878	24.3902	0	0.305	0.3	No		10.0	16.9	24.1	0.0	0.0	Dwg. 131-20020-00-C-2001 and 2002
SUBTOTAL - Mill Area Complex												**1532**	**2581**	**3677**	**4732**	**0**	
Mill Valley Fill Laydown																	
1	Reagent storage area	Strip footings		Bury-in-place	25	18	0	0.3	0.3	No		7.7	13.0	18.6	0.0	0.0	AutoCAD/Photos
SUBTOTAL - Mill Valley Fill Laydown												**8**	**13**	**19**	**0**	**0**	
Pelly Laydown																	
1	Pelly Laydown	Washpad		Bury-in-place	15	5	0.3	0	0	No		22.5	37.9	54.0	0.0	0.0	Photos
SUBTOTAL - Pelly Laydown												**23**	**38**	**54**	**0**	**0**	
Water Treatment Plant																	
1	Water Treatment Plant	No foundation present															
SUBTOTAL - Water Treatment Plant												**0**	**0**	**0**	**0**	**0**	

Notes:
1. Regrading of the areas are included in the elsewhere in the estimate.

C: Other Demolition

				Dimensions									Disposal	Demolition Quantities (Manual Calcs)				Transport			
No.	Area	Sub Area	Sub-component/ Building/ Section	Material	Quantity	Length (m)	Width (m)	Area (m^2)	Height/ Thickness (m)	Diameter (m)	Volume (m^3)	Air Factor	Disposal Location	Steel (tonnes)	Concrete In-place volume (m3)	Concrete Debris Volume (LCM)	Misc. Demolition Volume (m3)	Total Mass (tonnes)	Off-site Demolition Debris Trips	On-site Demolition Debris Volume (LCM)	Comments/Notes
Airstrip Area																					
1	Airstrip area	Exploration camp	misc. debris	Wood/Mixed Debris							50	1	On-site				50.0	35.0	0.0	50.0	Allowance based on photos
2		Waste Storage Areas	misc. debris	Wood/Mixed Debris							100	1	On-site				100.0	70.0	0.0	100.0	Allowance based on photos
3			Special waste pole barn	Wood/Mixed Debris		55	5		2.5		687.5	0.33	On-site				226.9	158.8	0.0	226.9	
4				Wood/Mixed Debris		32	2		2.5		160	0.33	On-site				52.8	37.0	0.0	52.8	
5		Storage East of strip	misc. debris	Wood/Mixed Debris							100	1	On-site				100.0	70.0	0.0	100.0	Allowance based on photos
SUBTOTAL - Airstrip Area														**0**	**0**	**0**	**530**	**371**	**0**	**530**	
Camp Area																					
1	Camp area	Utilidors	Mill to Camp	Wood/Mixed Debris		229	1		1		229	0.5	On-site				114.5	80.2	0.0	114.5	
2			East side of camp	Wood/Mixed Debris		181	1		1		181	0.5	On-site				90.5	63.4	0.0	90.5	
3		Covered walkways	Camp to Selkirk towers	Wood/Mixed Debris		55	2		2.5		275	0.33	On-site				90.8	63.5	0.0	90.8	
4			Misc. walkway/corridor allowa	Wood/Mixed Debris		100	2		2.5		500	0.33	On-site				165.0	115.5	0.0	165.0	
5		Smokers area	Building	Wood/Mixed Debris		2.5	2.5		2.5		15.625	0.33	On-site				5.2	3.6	0.0	5.2	
SUBTOTAL - Camp Area														**0**	**0**	**0**	**466**	**326**	**0**	**466**	
Fuel Farm Area																					
1	Fuel Farm Area	Generators	Generator wooden support st	Wood/Mixed Debris	8	6	3		1		18	0.33	On-site				5.9	10.0	0.0	14.3	AutoCAD/Photos
			Misc. piping/electrical conduit	Wood/Mixed Debris									On-site				15.0	25.3	0.0	36.1	Allowance
SUBTOTAL - Fuel Farm Area														**0**	**0**	**0**	**21**	**35**	**0**	**50**	
Mill Area Complex																					
1	Crushing & Ore Storage Re	Reclaim Tunnel	Feeder floor, stairs, ladders	Steel									On-site	7				6.8	0.0	9.7	Source Hatch: FS (2006)
2			Grating feeder floor	Steel	1			42	0.01				On-site	3				3.4	0.0	4.8	Source Hatch: FS (2006)
3			Mechanical (liners, conveyor	Steel									On-site	28				28.2	0.0	40.3	Source Hatch: FS (2006)
4		Ore Dump	Concrete components	Concrete	1						44.5		Bury-in-place		44.5	75		0.0	0.0	0.0	Source Hatch: FS (2006)
5		Reclaim area	Concrete components	Concrete	1						329		Bury-in-place		329.0	554		0.0	0.0	0.0	Source Hatch: FS (2006)
6	Mill Area	Mill Building	Process building elevated sla	Concrete									Bury-in-place		96.9	163		0.0	0.0	0.0	Source Hatch: FS (2006)
7			Structural steel platforms	Steel									On-site	108				108.0	0.0	154.3	Source Hatch: FS (2006)
8			Grating	Steel				527.9	0.01				On-site	42				42.5	0.0	60.7	Source Hatch: FS (2006)
9			Q-deck	Steel				690.3	0.001				On-site	6				5.6	0.0	7.9	Source Hatch: FS (2006)
10			Partition walls	Wood/Mixed Debris				130	0.15			0.5	On-site				9.8	16.4	0.0	23.5	Source Hatch: FS (2006)
11			Concrete separation walls	Concrete				408	0.15			1.3	On-site		61.2	79.6		113.4	0.0	162.0	Source Hatch: FS (2006)
12		Concentrate storage area	Coverall structure	Wood/Mixed Debris				2044	0.005		10.21925	10	On-site				102.2	172.1	0.0	245.9	Source Hatch: FS (2006)
13			Metal trusses	Steel	12	39.6341				0.3	33.6	2	On-site				67.2	113.2	0.0	161.8	Estimated based on dwg and photo
14		Grinding circuit	Misc. grinding ancillary equip	Steel	1								On-site	0.7				0.7	0.0	1.0	Source Hatch: FS (2006)
15			Process water tank	Steel	1								On-site	14.0				14.0	0.0	20.0	Source Hatch: FS (2006)
16			Fresh water tank	Steel	1								On-site	4.0				4.0	0.0	5.7	Source Hatch: FS (2006)
17		Flotation circuit	Misc. mechanical equip. stee	Steel									On-site	3.0				3.0	0.0	4.3	Source Hatch: FS (2006) (mostly pump frames
18		Concentrate dewatering	Concentrate Thickener	Steel	1			70.08312	2.13414634	6.097561	2.102494		On-site	16.9				16.9	0.0	24.2	Source Hatch: FS (2006)
19			Misc. mechanical equip. stee	Steel									On-site	5.2				5.2	0.0	7.4	Source Hatch: FS (2006)
20		Tailings Filter building	Concrete components	Concrete	1						313		Bury-in-place		44.5	75		0.0	0.0	0.0	Source Hatch: FS (2006)
21			Filter support steel and platfo	Steel									On-site	102.0				102.0	0.0	145.7	Source Hatch: FS (2006)
22			Q-deck	Steel				28	0.01				On-site	2				2.3	0.0	3.2	Source Hatch: FS (2006)
23			Concrete separation walls	Concrete				112	0.15			1.3	On-site		16.8	21.8		31.1	0.0	44.5	Source Hatch: FS (2006)
24			Architectural walls	Wood/Mixed Debris				100	0.1			0.5	On-site				5.0	8.4	0.0	12.0	Source Hatch: FS (2006)
25			Tanks & misc. mechanical eq	Steel									On-site	36				36.1	0.0	51.5	Source Hatch: FS (2006)
26			Thickener	Steel				205.8692	4.88	9.15	6.176075		On-site	49.7				49.7	0.0	71.0	
27		Reagents	Tanks & misc. mechanical eq	Steel									On-site	1				1.5	0.0	2.1	Source Hatch: FS (2006)
28		Mill building - phase 2	Interior retaining wall	Concrete									Bury-in-place		8.8	15		0.0	0.0	0.0	Source Hatch: FS (2006)
29			Structural steel platforms	Steel									On-site	37.1				37.1	0.0	53.0	Source Hatch: FS (2006)
30			Grating	Steel				696.8	0.01				On-site	56				56.1	0.0	80.1	Source Hatch: FS (2006)
31			Misc. mechanical equip. stee	Steel									On-site	1				1.0	0.0	1.4	Source Hatch: FS (2006)
32			Misc. mechanical equip. stee	Steel									On-site	2				2.4	0.0	3.4	Source Hatch: FS (2006)
33		Electricians shop	Entire building	Wood/Mixed Debris	1	12	10		1.5		180	0.33	On-site				59.4	100.0	0.0	142.9	
SUBTOTAL - Mill Area Complex														**526**	**602**	**983**	**244**	**1081**	**0**	**1544**	

Minto South Portal																				
1	Minto South Portal	Pad A	Misc. small storage areas/co	Wood/Mixed Debris	3	6	3		2		36	0.33	On-site				35.6	24.9	0.0	35.6	Allowance based on photos
2		Pad B	Green tents	Wood/Mixed Debris	2	18	12		5		1080	0.025	On-site				54.0	37.8	0.0	54.0	AutoCAD/Photos
3			Culverts/Debris	Wood/Mixed Debris	1	10	16		1.5		240	0.33	On-site				79.2	55.4	0.0	79.2	Allowance based on photos
4		Pad C	Culverts/Debris	Wood/Mixed Debris	1	37	12		2		888	0.33	On-site				293.0	205.1	0.0	293.0	Allowance based on photos
SUBTOTAL - Pelly Laydown														**0**	**0**	**0**	**462**	**323**	**0**	**462**	
Pelly Laydown																					
1	Pelly Laydown	Misc. small storage areas/covers		Wood/Mixed Debris	1	27	12		1.5		486	0.33	On-site				160.4	112.3	0.0	160.4	Est. Allowance based on photos/AutoCAD
SUBTOTAL - Pelly Laydown														**0**	**0**	**0**	**160**	**112**	**0**	**160**	
Water Treatment Plant																					
1	Water Treatment Plant		Misc. wooden walkways etc.	Wood/Mixed Debris	1												15.0	10.5			Estimated based on site photos.
SUBTOTAL - Water Treatment Plant														**0**	**0**	**0**	**15**	**11**	**0**	**0**	

Notes:
1. Where noted, demolition quantities were taken directly from the Hatch 2006 feasibility study.

D: Equipment and Material Removal

				Dimensions/Measurements									Disassembly/Disposal Details					Calculated Quantities			Transport		
No.	Area	Sub Area	Sub-component/ Building/ Section	Quantity	Length (m)	Width (m)	Height (m)	Diameter (m)	Volume (m3)	Surface Area (m2)	Volume (m3)	Mass of Item (tonnes)	Disposal Location	Requires washing/de contaminati on?	Crew Size	OVERRIDE - Time to dismantle (hrs)	OVERRIDE - Off-site number of trips	Small Disassembly Crew Time (hrs)	Large Disassembl y Crew Time (hrs)	Decontamination Time (hrs)	Off-site Removal Trips	On-site Demolition Debris Volume (LCM)	Comments/Notes
Airport Laydown																							
1	Airport Laydown	Tires			30	3	2				180.00	90.0	Off-site	No	Small	0		0.0	0.0	0.0	1.9	0.0	
		Seacans		1	16.0								Off-site	No	Small	0	0	0.0	0.0	0.0	0.0	0.0	Assumed to be filled other materials and removed.
		Propane tank		1									Off-site	No	Small	2	1.00	2.0	0.0	0.0	1.0	0.0	
		Misc. equipment		1								50.0	On-site	No	Small	0		0.0	0.0	0.0	0.0	71.4	Allowance based on photos
																		0.0	0.0	0.0	0.0	0.0	
SUBTOTAL - Airport Laydown																		**2**	**0**	**0**	**3**	**71**	
Airstrip Area																							
1	Airstrip area	Meteorological station		1								0.3	Off-site	No	Small	3		3.0	0.0	0.0	0.01	0.0	
		Land treatment facility	Waste Oil tanker	1								40.0	Off-site	Yes	Small	2	1	2.0	0.0	2.9	1.0	0.0	Mass of item bumped up to increase the decont. Time
		Incinerators	Primary incinerator	1								1.0	Off-site	No	Small	1		1.0	0.0	0.0	0.0	0.0	mass assumed
			Secondary incinerator	1								1.0	Off-site	No	Small	1		1.0	0.0	0.0	0.0	0.0	mass assumed
		Special waste pole barn	Poles	1	30	5	1.5		225			157.5	Off-site	No	Small	0		0.0	0.0	0.0	3.3	0.0	Dimensions/mass assumed based on aerial photos/size of facility
																		0.0	0.0	0.0	0.0	0.0	
SUBTOTAL - Airstrip Area																		**7**	**0**	**3**	**4**	**0**	
Camp Area																							
1	Camp Area	Offices	Safety equipment	1									Off-site	No	Small	8	0.25	8.0	0.0	0.0	0.3	0.0	
			Computer/office equipment	1									Off-site	No	Small	16	0.25	16.0	0.0	0.0	0.3	0.0	
			communication equip.	1									Off-site	No	Small	16	0.25	16.0	0.0	0.0	0.3	0.0	
		Camp	kitchen equipment/freezers, e	1									Off-site	No	Small	8	0.25	8.0	0.0	0.0	0.3	0.0	
																		0.0	0.0	0.0	0.0	0.0	
SUBTOTAL - Camp Area																		**48**	**0**	**0**	**1**	**0**	
Explosives Plant and Storage Areas																							
1	Explosives Plant/Fueling ar	Dyno compound	Fuel tank	1									Off-site	No	Small	2	1	2.0	0.0	0.0	1.0	0.0	Photos
2			Misc. tanks/debris etc.									25.0	On-site	No	Small			33.3	0.0	0.0	0.0	35.7	Allowance based on photos
3			Vertical tank	1									Off-site	No	Small	2	1	2.0	0.0	0.0	1.0	0.0	Photos
4			Explosive Storage	3	12	4	2		96				Off-site	No	Small	0	3	0.0	0.0	0.0	3.0	0.0	
																		0.0	0.0	0.0	0.0	0.0	
SUBTOTAL - Explosives Plant and Storage Areas																		**37**	**0**	**0.0**	**5**	**36**	
Fuel Farm Area																							
1	Fuel Farm Area	Generators	Generators	8	5	3	2.5		37.5				Off-site	No	Small	16	2	16.0	0.0	0.0	2.0	0.0	
2		Fueling Area	Northern Steel fuel tanks	4									Off-site	No	Small	4	4	4.0	0.0	0.0	4.0	0.0	
3			Misc. fuel lines/power/piping									2.0	On-site	Yes	Small	4		4.0	0.0	0.1	0.0	2.9	Allowance/photos
4			Diesel tank in NE corner of fa	1			6	3.5					Off-site	No	Large	4	1.00	0.0	4.0	0.0	1.0	0.0	
5			Large tanks	2			12.00	14.5		876.89705	9	141.2	On-site	Yes	Large			0.0	188.2	10.1	0.0	201.7	
SUBTOTAL - Fuel Farm Area																		**24**	**192**	**10.3**	**7**	**205**	
Mill Area Complex																							
1	Crushing & Ore Storage Re	Mobile Crusher/Feeder		1									Off-site	Yes	Small	10	1	10.0	0.0	1.0	1.0	0.0	Assume 1 day to prep for transport
2		Reclaim apron feeder			7.3	0.91	2.44						Off-site	Yes	Small	2	1	2.0	0.0	1.0	1.0	0.0	Dimensions form dwg. 1216002-R1
3		Slewing-Stockpile feed conv	Conveyor 36"x107 feet 40 h	1	51.8	1.0	4.39		227				Off-site	Yes	Large	30	4.2	0.0	30.0	1.0	4.2	0.0	Dimensions form dwg. 1216001-R1
4		SAG Mill Feed	Weightometer	1	3	1	0.6		2				Off-site	Yes	Small	2	0.19	2.0	0.0	1.0	0.2	0.0	Dimensions assumed
5		SAG Mill Feed	Conveyor, 36" x 130 feet	1	39.6	1.0	4.39		174				Off-site	Yes	Large	30	3.5	0.0	30.0	1.0	3.5	0.0	Dimensions form Hatch 2006
6	Mill Building - Phase 1	Mechanical	Small pumps (fire water)	3									Off-site	Yes	Small	1	0.0	1.0	0.0	1.0	0.0	0.0	Hatch (2006)
7		Mechanical	Fire water tank									14.0	On-site	No	Small			18.7	0.0	0.0	0.0	20.0	Hatch (2006)
8		Mechanical	O/H Crane	1									Off-site	Yes	Large	1	0.0	0.0	1.0	0.0	0.0	0.0	Hatch (2006)
9		Mechanical	Dust collector (20HP)	1									Off-site	Yes	Small	1	0.0	1.0	0.0	1.0	0.0	0.0	Hatch (2006)
10	Grinding Circuit - Phase 1	SAG Mill	Sag Mill	1				5.030488				50.0	Off-site	Yes	Large		3.0	0.0	66.7	3.6	3.0	0.0	Scaled from previous project based on diameter
11		Ball Mill Phase 1	10.5' x 12'	1	3.2	3.7						60.0	Off-site	Yes	Large		2.0	0.0	80.0	4.3	2.0	0.0	Scaled from previous project based on diameter
12		Grinding ancillary equip.	Pump box	1								2.0	Off-site	Yes	Small			2.7	0.0	0.1	0.0	0.0	
13			Rubber lining - 6.5'x6.5'x10'	1						27		0.2	On-site	Yes	Small			0.3	0.0	0.0	0.0	0.3	
14			Discharge chutes	2								2.6	On-site	Yes	Small			3.5	0.0	0.2	0.0	3.7	
15			Cyclo-pac 3x15" dia.	1	3.0	4.6	3.05		42.51			8.6	Off-site	Yes	Large			0.0	11.4	0.6	0.2	0.0	pick-up truck factor applied'
16			Cyclone feed pumps	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
17			Sump pump #1	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
18			Pump 8"x6' process water pu	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
19			Pump 1" x 1.5" fresh water pu	2								2.0	Off-site	No	Small	2		2.0	0.0	0.0	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
20	Flotation Circuit - Phase 1	Flotation mechanical equip.	Reclaim/conc feed pump	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
21			Flotation Area 1 sump pump	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
22			Floatation cells 15m3	5			3.00	2.523133	15.00	23.779971	0	9.6	On-site	Yes	Large			0.0	12.8	0.7	0.0	13.7	Assumed dimensions and steel thickness
23			Flotation cells (2.8m3)	6			2.00	1.335118	2.80	8.3887937	0	4.1	On-site	Yes	Small			5.4	0.0	0.3	0.0	5.8	
24			Rougher concentrate pump 1	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
25			Concentrate thickener feed p	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
26			Flotation blowers	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
27			Hoist	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
28	Concentrate Dewatering	Mechanical equip.	Filter, 30m2 40 hp	1	4.6	3.7	4.57				77	15.4	Off-site	Yes	Large	16		0.0	16.0	1.1	0.3	0.0	pick-up truck factor applied'
29			Concentrate discharge conve	1	50.0	0.6	2.44						Off-site	Yes	Large	16	2.1	0.0	16.0	1.0	2.1	0.0	
30			Concentrate thickener OF pu	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
31			Concentrate filter feed pump	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
32			Sump pump, 2" vert 5HP	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
33			Truck scale	1									Off-site	Yes	Large	4	1.0	0.0	4.0	1.0	1.0	0.0	
34	Tailings Filter Building	Mechanical equip.	Tailings pump	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
35			Process water tank feed pum	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
36			Tailings U/F Pumps	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
37			Belt filter feed pumps	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
38			Belt filter	2	11.0	3.0	6.10						Off-site	Yes	Large	16	2.0	0.0	16.0	1.0	2.0	0.0	

39			Concentrate discharge conve	1	50.0	1.0	2.44						Off-site	Yes	Large	16	2.1	0.0	16.0	1.0	2.1	0.0	
40			Concentrate stockpile feed co	1	50.0	1.0	2.44						Off-site	Yes	Large	16	2.1	0.0	16.0	1.0	2.1	0.0	
41			Haul truck freed conveyor	1	92.3	1.0	2.44						Off-site	Yes	Large	16	3.4	0.0	16.0	1.0	3.4	0.0	
42			Fresh water pumps	2								2.0	Off-site	No	Small	2		2.0	0.0	0.0	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
43			Wash water pumps	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
44			Vacuum pump #1	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
45			Tailings filtrate pumps	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
46			Filter area sump pump	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
47			Tailings area sump pump	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
48			Belt filter compressors	2	3.0	3.0	2.44		22.67			9.1	Off-site	Yes	Large			0.0	12.2	0.7	0.2	0.0	
49			Tailings Area Boiler	1	4.6	1.5	1.22		8.50			1.7	Off-site	Yes	Small			2.3	0.0	0.1	0.0	0.0	
50			Boiler feed pumps	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
51			Boiler blower	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
52			Boiler fuel pumps	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
53			Water collection pumps	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
54	Reagents	Mechanical equip.	Metering pumps	11								11.0	Off-site	Yes	Small	11		11.0	0.0	0.8	0.2	0.0	assumed 1 hr for dismantling, 1 tonne each
55			Pax transfer pump	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
56			Thickener flocculent unit	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
57	Mill Services	Mechanical equip.	Reclaim water pump	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
58			Mill water Pump	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
59			Mill water Booster Pump	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
60			Plant air compressors	2	1.8	0.9	0.91		1.53			2.0	Off-site	Yes	Small	2	0.1	2.0	0.0	0.1	0.1	0.0	
61	First fills	Grinding balls	Ball Mill and SAG Mill									184.5	On-site	No	Small			246.0	0.0	0.0	0.0	263.6	
62	Mill Building - Phase 2	Mechanical equip.	O/H Crane Ball Mill area		22.9					0.0055914		1.5	Off-site	Yes	Large			0.0	2.0	0.1	0.0	0.0	Assumed I-beam dimensions for crane rail, plus 0.5 tonnes for crane
63	Grinding Circuit - Phase 2	Mechanical Grinding	Grinding area sump pump#2	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
64			Cyclo-pac 2x26' dia.	1	3.0	4.6	3.05		42.51			8.6	Off-site	Yes	Large			0.0	11.4	0.6	0.2	0.0	pick-up truck factor applied'
65		Grinding ancillary equip.	No. 2 floatation feed sampler	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
66			Pump box	1								2.0	Off-site	Yes	Small			2.7	0.0	0.1	0.0	0.0	
67			SAG discharge pump	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
68			VFD cyclone feed pump	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
69	Flotation Circuit - Phase 2	Mechanical equip.	Floatation feed pumps	2								2.0	Off-site	Yes	Small	2		2.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
70			Floatation cells 15m3	5			3.00	2.523133	15.00	23.779971	0	9.6	On-site	Yes	Large			0.0	12.8	0.7	0.0	13.7	Assumed dimensions and steel thickness
71			Pump box	1								2.0	Off-site	Yes	Small			2.7	0.0	0.1	0.0	0.0	
72			Sump pump	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
73			Recleaner tails pump	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
74			Rougher concentrate pump 2	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
75			Floatation cells 100 cu. Ft	4			2.00	1.335118	2.80	8.3887937	0	2.7	On-site	Yes	Small			3.6	0.0	0.2	0.0	3.9	
76	Tailings Deposal - phase 2	Mechanical equip.	Belt filter #3 (90m2 x 3m)	1	18.3	3.0	6.10		334.62			67.3	Off-site	Yes	Large		3.0	0.0	89.8	4.8	3.0	0.0	dwg. 35-10-105; 'pick-up truck factor applied)
77			Fresh water pumps	1								1.0	Off-site	No	Small	1		1.0	0.0	0.0	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
78			Wash water pumps	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
79			Vacuum pump #3	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
80			Tailings filtrate pumps	1								1.0	Off-site	Yes	Small	1		1.0	0.0	0.1	0.0	0.0	assumed 1 hr for dismantling, 1 tonne each
81			Belt filter compressor	1	3.0	3.0	2.44		22.67			4.6	Off-site	Yes	Large			0.0	6.1	0.3	0.1	0.0	pick-up truck factor applied'
SUBTOTAL - Mill Area Complex																		**379**	**466**	**38**	**33**	**325**	
Mill Valley Fill Laydown																							
1	MVF Laydown Area	Grinding media											On-site	No	Small			0.0	0.0	0.0	0.0	0.0	
2		Steel rack	misc.		76.0	2.00	1.00		152			106.4	On-site	No	Small	0		0.0	0.0	0.0	0.0	152.0	
3		Tire racks	Tires		150.0	3.0	1.50		675			472.5	Off-site	No	Small	0		0.0	0.0	0.0	9.8	0.0	
4		Site Services			100	2	1		200			140.0	On-site	No	Small	0		0.0	0.0	0.0	0.0	200.0	
5		Mill Elect			40	2	2.5		200			140.0	On-site	No	Small	0		0.0	0.0	0.0	0.0	200.0	
6		Misc.			50	2	1		100			70.0	On-site	No	Small	0		0.0	0.0	0.0	0.0	100.0	
																		0.0	0.0	0.0	0.0	0.0	
SUBTOTAL - Mill Valley Fill Laydown																		**0**	**0**	**0**	**10**	**652**	
Minto South Portal																							
1	Minto South Portal	Pad A	Fuel tank	1									Off-site	No	Small	2	1	2.0	0.0	0.0	1.0	0.0	
		Pad C	Propane Fuel tanks	2									Off-site	No	Small	4	2	4.0	0.0	0.0	2.0	0.0	
																		0.0	0.0	0.0	0.0	0.0	
																		0.0	0.0	0.0	0.0	0.0	
SUBTOTAL - Minto South Portal																		**6**	**0**	**0**	**3**	**0**	
Pelly Laydown																							
1	Pelly Laydown	Tires			30	3	2				180.00	90.0	Off-site	No	Small	0		0.0	0.0	0.0	1.9	0.0	
		Seacans/trailers		7	16.0								Off-site	No	Small	0	0	0.0	0.0	0.0	0.0	0.0	Assumed to be filled other materials and removed.
		Misc. equipment		1								50.0	On-site	No	Small	0		0.0	0.0	0.0	0.0	71.4	Allowance based on photos
																		0.0	0.0	0.0	0.0	0.0	
SUBTOTAL - Pelly Laydown																		**0**	**0**	**0**	**2**	**71**	
Vent Raises - Area 118																							
1	Area 118	Air Heater	All components	1								44.6	Off-site	No	Small	20	1	20.0	0.0	0.0	1.0	0.0	Dwg. 9057-MGD; Mass of equipment listed on drawings.
2																							
SUBTOTAL - Vent Raises - Area 118																		**20**	**0**	**0**	**1**	**0**	
Water Treatment Plant																							
1	Water Treatment Plant	WTP building	Actiflo clarifier	1	6.85976	2.7439	3.04878		57.39			11.5	Off-site	No	Large		1	0.0	15.4	0.0	1.0	0.0	Dwg. MIN-000-GA-01 and-02; pick-up truck factor applied
2			BQE skid #1	1	9.1	2.62	1.52		36.56			1.5	Off-site	No	Small		1	2.0	0.0	0.0	1.0	0.0	Dwg. MIN-000-GA-01 and-02
3			BQE Skid #2	1	4.6	2.13	2.22		22.05			1.5	Off-site	No	Small		0.5	2.0	0.0	0.0	0.5	0.0	Dwg. MIN-000-GA-01 and-02
4			BQE Skid #3	1	5.9	3.57	2.44		51.75			10.4	Off-site	No	Large		1	0.0	13.9	0.0	1.0	0.0	Dwg. MIN-000-GA-01 and-02
5			BQE Skid #4	1	8.2	3.05	0.91		22.95			1.0	Off-site	No	Small		0.5	1.3	0.0	0.0	0.5	0.0	Dwg. MIN-000-GA-01 and-02
6			RO Skids	2	7.8	1.83	1.50		21.33			8.6	Off-site	No	Small		0.5	11.4	0.0	0.0	0.5	0.0	Dwg. MIN-000-GA-01 and-02
SUBTOTAL - Water Treatment Plant																		**17**	**29**	**0**	**5**	**0**	

Notes:
1. Assumes steel material
2. See disassembly crew details below
3. Time to dismantle includes time to prepare for transport
4. Where noted, demolition quantities were taken directly from the Hatch 2006 feasibility study.
5. HVAC equipment (fans, ducts, heater, etc.) assumed to be part of demo debris and considered as part of the wall thickness in Part A.
6. 'pick-p truck factor' = 20 m3 bulk volume ~ 4 tonnes

Worksheet 24 - Mobilization-Demobilization Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

A: Equipment Mobilization - Year 0

Equipment

		Equipment Input				Breakdown and Assembly Time (each)			Transport Time (each)						Calculations									
Equip. Code	Equipment	Qnty	Shipping Weight (tonne)	Transport Vehicle 1 (axles)	Transport Vehicle #2 (axles)	Breakdown Time (hrs)	Assembly Time (hrs)	Mechanic Workforce Size	Shipping Origin	Shipping Distance (km)	Average Speed to Barge (km/hr)	Average Speed from Barge to Site (km/hr)	Loading and Off-Loading Time (hrs)	# of Pilot Vehicles required per equip.	Meals/Hotel Allowance ($)	Total Mechanic Hourly Rate ($/hr)	Breakdown-Assembly Cost Per Equip.	Travel Time to Barge (hrs)	Travel Time Barge to Site (hrs)	Total Travel Time (1-way) (hrs)	Equipment Rate	Transport Vehicle 1 Hourly Rate	Transport Vehicle 2 Hourly Rate	Total Transport Cost per Equip.	Total cost Per Equip.	TOTAL COST	Comments/Notes
E.03.1	CAT D6R	1	18	5	0	4	4	3	Whitehorse	283	80	35	3	0	$50	$207.96	$1,664	4	1.1	5	$105.36	$83.95	$0.00	$1,909.14	$3,573	$3,573	
E.03.2	CAT D8R	2	38	8	0	4	8	3	Whitehorse	283	80	35	3	2	$50	$207.96	$2,496	4	1.1	5	$128.31	$110.02	$0.00	$3,308.87	$5,804	$11,609	
E.03.3	CAT D10T	2	66	8	5	4	8	3	Whitehorse	283	80	35	3	2	$100	$207.96	$2,496	4	1.1	5	$194.68	$110.02	$83.95	$4,897.18	$7,393	$14,785	
E.05.1	CAT 330	3	35	7	0	4	4	3	Whitehorse	283	80	35	3	0	$50	$207.96	$1,664	4	1.1	5	$83.65	$107.46	$0.00	$2,032.04	$3,696	$11,087	
E.05.2	CAT 385	1	83	8	8	8	16	3	Whitehorse	283	80	35	3	2	$100	$207.96	$4,991	4	1.1	5	$177.68	$110.02	$110.02	$5,087.52	$10,079	$10,079	
E.06.2	CAT 16M (Class 7 - 250-300 FWHP)	1	25	6	0	4	4	3	Whitehorse	283	80	35	3	0	$50	$207.96	$1,664	4	1.1	5	$83.88	$92.87	$0.00	$1,854.60	$3,518	$3,518	
E.07.2	CAT 735	5	30	6	0	4	4	3	Whitehorse	283	80	35	3	0	$50	$207.96	$1,664	4	1.1	5	$89.43	$92.87	$0.00	$1,897.06	$3,561	$17,804	
E.07.4	CAT 773G	5	45	8	5	4	12	3	Whitehorse	283	80	35	3	2	$100	$207.96	$3,327	4	1.1	5	$142.14	$110.02	$83.95	$4,495.72	$7,823	$39,116	
E.08.1	Hydraulic Crane, 30Ton	1	30	7	0	4	4	3	Whitehorse	283	80	35	3	2	$100	$207.96	$1,664	4	1.1	5	$81.38	$107.46	$0.00	$2,968.77	$4,632	$4,632	
E.09.1	CAT 950H	1	18	5	0	4	4	3	Whitehorse	283	80	35	3	0	$100	$207.96	$1,664	4	1.1	5	$61.19	$83.95	$0.00	$1,621.64	$3,285	$3,285	
E.09.2	CAT 990	1	78	8	8	8	16	3	Whitehorse	283	80	35	3	2	$100	$207.96	$4,991	4	1.1	5	$104.53	$110.02	$110.02	$4,528.55	$9,520	$9,520	
E.10.2	Fuel Truck	1	n/a	0	0	0	0	0	Whitehorse	283	80	60	3	0	$100	$0.00	$0	4	0.8	4	$3.86	$44.84	$0.00	$672.09	$672	$672	
E.10.4	Highway Service/Flatbed Truck (HIAB) 5 ton	1	n/a	0	0	0	0	0	Whitehorse	283	80	60	3	0	$100	$0.00	$0	4	0.8	4	$20.79	$44.84	$0.00	$796.02	$796	$796	
E.10.12	Water Truck (5,000 gal)	1	n/a	0	0	0	0	0	Whitehorse	283	80	60	3	0	$100	$0.00	$0	4	0.8	4	$15.49	$44.84	$0.00	$757.21	$757	$757	
E.11.2	Boomlift: 60' articulating	1	10	2	0	0	0	0	Whitehorse	283	80	60	3	0	$100	$0.00	$0	4	0.8	4	$25.00	$36.07	$0.00	$724.81	$725	$725	
E.11.4	Bus: Passenger Bus (24 person cap.)	1	n/a	0	0	0	0	0	Whitehorse	283	80	50	3	0	$100	$0.00	$0	4	0.9	4	$75.30	$43.84	$0.00	$1,198.31	$1,198	$1,198	
E.11.7	Crusher (200 Tons/hr)	1	25	8	8	8	16	3	Whitehorse	283	80	35	3	2	$100	$207.96	$4,991	4	1.1	5	$170.00	$110.02	$110.02	$5,028.82	$10,020	$10,020	
E.11.10	Emergency Transport Vehicle	1	n/a	0	0	0	0	0	Whitehorse	283	80	60	0	0	$0	$0.00	$0	4	0.8	4	$150.30	$43.84	$0.00	$1,028.06	$1,028	$1,028	
E.11.18	Screener (finley 583) (~400tonnes/hr)	1	75	8	8	8	16	3	Whitehorse	283	80	35	3	2	$100	$207.96	$4,991	4	1.1	5	$189.59	$110.02	$110.02	$5,178.50	$10,170	$10,170	
n/a	Misc. Equipment/Supplies	4	#N/A	8	0	0	0	0	Whitehorse	283	80	40	3	0	$100	$0.00	$0	4	1.0	5	$0.00	$110.02	$0.00	$1,452.78	$1,453	$5,811	
TOTAL - Equipment Mobilization Costs																										**$162,039**	

Notes:
1. See Transport vehicle details below for number capacity.
2. Loading and off-loading time is a total (ie. Include both loading and unloading).
3. Calculations include costs for transport vehicles to return to shipping origin.
4. Mob costs for materials are included in the material rates.

B: Equipment Demobilization - End of Post-Closure 2

1. Demobilization assumed to consist of the same equipment assumed in the Year 0 mobilization, EXCEPT for a reduced fleet that is assumed to remain for post closure maintenance/repairs activities.

Equip. Code	Equipment to remain during post closure	Qnty	TOTAL MOB COST (from Part A)
E.03.1	CAT D6R	1	$3,573
E.05.1	CAT 330	1	$3,696
E.07.2	CAT 735	2	$3,561
E.09.1	CAT 950H	1	$3,285
E.06.2	CAT 16M (Class 7 - 250-300 FWHP)	1	$3,518
TOTAL:			**$17,633**

TOTAL Equipment Demobilization Cost at end of Active Closure: **$144,406**

C: Equipment Mobilization - Perpetual Care Year

Equipment

		Equipment Input				Breakdown and Assembly Time (each)			Transport Time (each)						Calculations									
Equip. Code	Equipment	Qnty	Shipping Weight (kg)	Transport Vehicle 1 (axles)	Transport Vehicle #2 (axles)	Breakdown Time (hrs)	Assembly Time (hrs)	Mechanic Workforce Size	Shipping Origin	Shipping Distance (km)	Average Speed to Barge (km/hr)	Average Speed from Barge to Site (km/hr)	Loading and Off-Loading Time (hrs)	# of Pilot Vehicles required per equip.	Meals/Hotel Allowance ($)	Total Mechanic Hourly Rate ($/hr)	Breakdown-Assembly Cost Per Equip.	Travel Time to Barge (hrs)	Travel Time Barge to Site (hrs)	Total Travel Time (1-way) (hrs)	Equipment Rate	Transport Vehicle 1 Hourly Rate	Transport Vehicle 2 Hourly Rate	Total Transport Cost per Equip.	Total cost Per Equip.	TOTAL COST	Comments/Notes
E.03.1	CAT D6R	1	18	5	0	4	4	3	Whitehorse	283	80	35	3	0	$50	$207.96	$1,664	4	1.1	5	$105.36	$83.95	$0.00	$1,909.14	$3,573	$3,573	
E.05.1	CAT 330	1	35	7	0	4	4	3	Whitehorse	283	80	35	3	0	$50	$207.96	$1,664	4	1.1	5	$83.65	$107.46	$0.00	$2,032.04	$3,696	$3,696	
E.07.2	CAT 735	2	30	6	0	4	4	3	Whitehorse	283	80	35	3	0	$50	$207.96	$1,664	4	1.1	5	$89.43	$92.87	$0.00	$1,897.06	$3,561	$7,122	
E.09.1	CAT 950H	1	18	5	0	4	4	3	Whitehorse	283	80	35	3	0	$50	$207.96	$1,664	4	1.1	5	$61.19	$83.95	$0.00	$1,571.64	$3,235	$3,235	
E.10.4	Highway Service/Flatbed Truck (HIAB) 5 ton	1	n/a	0	0	0	0	0	Whitehorse	283	80	60	0	0	$50	$0.00	$0	4	0.8	4	$20.79	$44.84	$0.00	$527.19	$527	$527	
n/a	**MISC. Equipment**	1	#N/A	8	0	0	0	0	Whitehorse	283	80	40	3	0	$50	$0.00	$0	4	1.0	5	$0.00	$110.02	$0.00	$1,402.78	$1,403	$1,403	Fuel tanks, etc.
TOTAL - Equipment Mobilization Costs																										$19,555	

Notes:
1. See Transport vehicle details below for number capacity.
2. Loading and off-loading time is a total (i.e.. Include both loading and unloading).
3. Calculations include costs for transport vehicles to return to shipping origin.

Worksheet 25 - Monitoring Costs

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Subr February 6, 2018
File Locatior \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

Summary of Monitoring Costs Per Event

		Annual Cost Per Phase						
		1 - Interim	2 - Active	3 - PC 1	4 - PC 2	5 - Perpetual	n/a	Comments/Notes
1	Water Quality Monitoring	$115,295	$130,437	$51,987	$27,220	$27,664	$0	
2	Sediment Monitoring	$3,388	$3,388	$3,388	$3,388	$3,388	$0	Assumes samples collected during water quality sampling.
3	Biological Monitoring	$12,294	$12,294	$12,294	$12,294	$12,294	$0	
4	Geotechnical Monitoring	$12,409	$12,409	$12,409	$23,216	$23,216	$0	
5	Revegetation Monitoring	$0	$10,530	$10,530	$10,530	$10,530	$0	
6	Annual Inspection Reporting	$2,372	$2,372	$2,372	$23,900	$23,900	$0	

Monitoring Schedule and Total Cost Summary

1. Year 0 and Peak Scenario

		# of Years						
		1 - Interim	2 - Active	3 - PC 1	4 - PC 2	5 - Perpetual	TOTAL COST	Comments/Notes
1	Water Quality Monitoring	2	3	5	10	9	$1,403,014	
2	Sediment Monitoring	2	3	3	2	9	$64,372	
3	Biological Monitoring	2	3	3	2	9	$233,587	
4	Geotechnical Monitoring	2	3	5	2	9	$379,471	
5	Revegetation Monitoring	0	3	5	2	9	$200,071	
6	Annual Inspection Reporting	2	3	5	2	9	$286,620	

2. EOM Scenario

		# of Years						
		1 - Interim	2 - Active	3 - PC 1	4 - PC 2	5 - Perpetual	TOTAL COST	Comments/Notes
1	Water Quality Monitoring	0	3	5	10	9	$1,172,423	
2	Sediment Monitoring	0	3	3	2	9	$57,596	
3	Biological Monitoring	0	3	3	2	9	$208,999	
4	Geotechnical Monitoring	0	3	5	2	9	$354,653	
5	Revegetation Monitoring	0	3	5	2	9	$200,071	
6	Annual Inspection Reporting	0	3	5	2	9	$281,876	

NOTES:
1. These tables are used as a check to the NPV calculation spreadsheets.

Phase 1 - Annual Monitoring Costs - Prior to Closure

						Unit Rates			Activity Totals					
WBS	Facility/Area	Task	Crew/ Qnty	Unit	Hours	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Material Cost	Labour Cost	Equipment Cost	Cost	Sub-Totals	Source / Comments
M1.1	**Water Quality Monitoring**												**$115,295**	
M1.1.1		Collect water quality sampling (by onsite staff)	2	ea	144		$45	$12		$12,837	$1,778	$14,615		
M1.1.2		Complete lab testing	640	yr		$150			$96,000			$96,000		test unit rate based on actual site costs.
M1.1.3		Reporting	1	ea	40		$117			$4,680		$4,680		
M1.2	**Sediment Monitoring**												**$3,388**	
M1.2.1		Collect data/samples	2	ea	0		$45	$12		$0.00	$0.00	$0		
M1.2.2		Laboratory Analysis	2	ea		$290			$580			$580		Test unit rate based on actual costs. (Total metals + contaminates)
M1.2.3		Reporting	1	ls	24		$117			$2,808		$2,808		
M1.3	**Biological Monitoring**												**$12,294**	
M1.3.1		Consultant travel - Whitehorse to/from site	2	ls	8.1	$50	$117		$100	$1,892		$1,992		
M1.3.2		Truck Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M1.3.3		Collect data/samples	2	ls	8		$117			$1,872		$1,872		
M1.3.4		Laboratory Analysis	10	ea		$270			$2,700			$2,700		Biological test actual costs range from $180 to $270 depending on test type
M1.3.5		Reporting	1	ls	40		$117			$4,680		$4,680		
M1.4	**Geotechnical Monitoring**												**$12,409**	
M1.4.1.1	Annual Geotech Inspection	Consultant travel - Whitehorse to/from site	1	ls	8.1	$50	$117		$50	$946		$996		
M1.4.1.2		Truck Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M1.4.1.3		Site Inspection	1	ls	12		$117			$1,404		$1,404		
M1.4.1.4		Reporting	1	ls	40		$117			$4,680		$4,680		
M1.4.2.1	Instrumentation Monitoring	Collect data, plot and review trends	1	ls	96		$45			$4,279		$4,279		
M1.5	**Annual Reclamation Reports**												**$2,372**	
M1.5.1		Reporting	1	ls	16	$500	$117		$500	$1,872		$2,372		
TOTAL												**$145,758**	**$145,758**	

NOTES:
1. Assumes water quality sampling completed by site personnel; Number of test obtained from Table 2-2 in EMSRP
2. Assumes sediment samples collected during water quality sampling.
3. Assumes biological sampling completed by consultants based in Whitehorse, and sampling takes place while barge in place.
4. Assumes geotechnical instrumentation readings collected and plotted by on-site staff.
5. Assumes annual reclamation report compiled by on-site staff whose salary is included in indirect costs; cost included here is for additional consultant review.

Phase 2 - Annual Monitoring Costs - Active Closure

						Unit Rates			Activity Totals					
WBS	Facility/Area	Task	Crew/ Qnty	Unit	Hours	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Material Cost	Labour Cost	Equipment Cost	Cost	Sub-Totals	Source / Comments
M2.1	**Water Quality Monitoring**												**$130,437**	
M2.1.1	Sampling during camp ops.	Collect water quality sampling (by onsite staff)	2	ea	42		$45	$12		$3,744	$1,037.37	$4,781		
M2.2.1	Winter sampling	Helicopter time (5 months)	5	ea	2.25		$49	$1,600		$546	$18,000	$18,546		Helicopter cost based on onsite costs
M2.2.2		Consultant time incl. travel	5	ea	8		$117			$4,680		$4,680		1 consultant
M2.2.3		Car Rental (inc. fuel)	5	days				$350			$1,750	$1,750		
M2.3.1	Lab testing	Complete lab testing	640	ea		$150			$96,000			$96,000		
M2.3.2		Reporting	1	ea	40		$117			$4,680		$4,680		
M2.2	**Sediment Monitoring**												**$3,388**	
M2.2.1		Collect data/samples	2	ea	0		$45	$12		$0	$0	$0		
M2.2.2		Laboratory Analysis	2	ea		$290			$580			$580		
M2.2.3		Reporting	1	ls	24		$117			$2,808		$2,808		
M2.3	**Biological Monitoring**												**$12,294**	
M2.3.1		Consultant travel - Whitehorse to/from site	2	ls	8.1	$50	$117		$100	$1,892		$1,992		
M2.3.2		Truck Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M2.3.3		Collect data/samples	2	ls	8		$117			$1,872		$1,872		
M2.3.4		Laboratory Analysis	10	ea		$270			$2,700			$2,700		
M2.3.5		Reporting	1	ls	40		$117			$4,680		$4,680		
M2.4	**Geotechnical Monitoring**												**$12,409**	
M2.4.1.1	Annual Geotech Inspection	Consultant travel - Whitehorse to/from site	1	ls	8.1	$50	$117		$50	$946		$996		
M2.4.1.2		Truck Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M2.4.1.3		Site Inspection	1	ls	12		$117			$1,404		$1,404		
M2.4.1.4		Reporting	1	ls	40		$117			$4,680		$4,680		
M2.4.2.1	Instrumentation Monitoring	Collect data, plot and review trends	1	ls	96		$45			$4,279		$4,279		
M2.5	**Reclamation Monitoring**												**$10,530**	
M2.5.1		Consultant travel - Whitehorse to/from site	2	ls	8.1	$50	$117		$100	$1,892		$1,992		
M2.5.2		Truck Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M2.5.3		Inspect vegetation	2	ls	12		$117			$2,808		$2,808		
M2.5.4		Reporting	1	ls	40		$117			$4,680		$4,680		
M2.6	**Annual Reclamation Reports**												**$2,372**	
M2.6.1		Reporting	1	ls	16	$500	$117		$500	$1,872		$2,372		
TOTAL												**$171,431**	**$171,431**	

NOTES:
1. Assumes water quality sampling completed by site personnel when camp in operation, and by helicopter during the winter months
2. Assumes sediment samples collected during water quality sampling.
3. Assumes biological sampling completed by consultants based in Whitehorse, and sampling takes place while barge in place.
4. Assumes geotechnical instrumentation readings collected and plotted by on-site staff.
5. Assumes annual reclamation report compiled by on-site staff whose salary is included in indirect costs; cost included here is for additional consultant review.

Phase 3 - Annual Monitoring Costs - Post-Closure 1

						Unit Rates			Activity Totals					
WBS	Facility/Area	Task	Crew/ Qnty	Unit	Hours	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Material Cost	Labour Cost	Equipment Cost	Cost	Sub-Totals	Source / Comments
M3.1	**Water Quality Monitoring**												**$51,987**	
M3.1.1.1	Sampling during camp ops.	Collect water quality samples	2	ea	42		$45	$12		$3,744	$1,037	$4,781		
M3.1.2.1	Winter sampling	Helicopter time (5 months)	5	ea	2.25		$49	$1,600		$546	$18,000	$18,546		Helicopter cost based on onsite costs
M3.1.2.2		Consultant time incl. travel	5	ea	8		$117			$4,680		$4,680		1 consultant
M3.1.2.3		Car Rental (inc. fuel)	5	days				$350			$1,750	$1,750		
M3.1.3.1	Lab testing	Complete lab testing	117	ea		$150			$17,550			$17,550		
M3.1.3.2		Reporting	1	ea	40		$117			$4,680		$4,680		
M3.2	**Sediment Monitoring (Bi - Annual)**												**$3,388**	
M3.2.1		Collect data/samples	2	ea	0		$45	$12		$0.00	$0.00	$0		Samples collected during water sampling
M3.2.2		Laboratory Analysis	2	ea		$290			$580			$580		
M3.2.3		Reporting	1	ls	24		$117			$2,808		$2,808		
M3.3	**Biological Monitoring (Bi-Annual)**												**$12,294**	
M3.3.1		Consultant travel - Whitehorse to/from site	2	ls	8.1	$50	$117		$100	$1,892		$1,992		
M3.3.2		Truck Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M3.3.3		Collect data/samples	2	ls	8		$117			$1,872		$1,872		
M3.3.4		Laboratory Analysis	10	ea		$270			$2,700			$2,700		
M3.3.5		Reporting	1	ls	40		$117			$4,680		$4,680		
M3.4	**Geotechnical Monitoring**												**$12,409**	
M3.4.1.1	Annual Geotech Inspection	Consultant travel - Whitehorse to/from site	1	ls	8.1	$50	$117		$50	$946		$996		
M3.4.1.2		Truck Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M3.4.1.3		Site Inspection	1	ls	12		$117			$1,404		$1,404		
M3.4.1.4		Reporting	1	ls	40		$117			$4,680		$4,680		
M3.4.2.1	Instrumentation Monitoring	Collect data, plot and review trends	1	ls	96		$45			$4,279		$4,279		
M3.5	**Reclamation Monitoring**												**$10,530**	
M3.5.1		Consultant travel - Whitehorse to/from site	2	ls	8.1	$50	$117		$100	$1,892		$1,992		
M3.5.2		Truck Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M3.5.3		Inspect vegetation	2	ls	12		$117			$2,808		$2,808		
M3.5.4		Reporting	1	ls	40		$117			$4,680		$4,680		
M3.6	**Annual Reclamation Reports**												**$2,372**	
M3.6.1		Reporting	1	ls	16	$500	$117		$500	$1,872		$2,372		
TOTAL												**$92,981**	**$92,981**	

NOTES:
1. Assumes water quality sampling completed by site personnel when camp in operation, and by helicopter during the winter months
2. Assumes sediment samples collected during water quality sampling.
3. Assumes biological sampling completed by consultants based in Whitehorse, and sampling takes place while barge in place.
4. Assumes geotechnical instrumentation readings collected and plotted by on-site staff.
5. Assumes annual reclamation report compiled by on-site staff whose salary is included in indirect costs; cost included here is for additional consultant review.

Phase 4 - Annual Monitoring Costs - Post-Closure 2

						Unit Rates			Activity Totals					
WBS	Facility/Area	Task	Crew/ Qnty	Unit	Hours	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Material Cost	Labour Cost	Equipment Cost	Cost	Sub-Totals	Source / Comments
M4.1	**Water Quality Monitoring**												**$27,220**	
M4.1.1		Helicopter time (3 rounds)	3	ea	2.25		$49	$1,600		$328	$10,800	$11,128		Helicopter cost based on onsite costs
M4.1.2		Consultant time incl. travel	3	ea	12		$117			$4,212		$4,212		1 consultant
M4.1.3		Car Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M4.1.4		Complete lab testing	41	ea		$150			$6,150			$6,150		
M4.1.5		Reporting	1	ea	40		$117			$4,680		$4,680		
M4.2	**Sediment Monitoring (Bi - Annual)**												**$3,388**	
M4.2.1		Collect data/samples	2	ea	0		$45	$12		$0.00	$0.00	$0		Samples collected during water sampling
M4.2.2		Laboratory Analysis	2	ea		$290			$580			$580		
M4.2.3		Reporting	1	ls	24		$117			$2,808		$2,808		
M4.3	**Biological Monitoring (Bi-Annual)**												**$12,294**	
M4.3.1		Consultant travel - Whitehorse to/from site	2	ls	8.1	$50	$117		$100	$1,892		$1,992		
M4.3.2		Truck Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M4.3.3		Collect data/samples	2	ls	8		$117		$0	$1,872		$1,872		
M4.3.4		Laboratory Analysis	10	ea		$270			$2,700			$2,700		
M4.3.5		Reporting	1	ls	40		$117			$4,680		$4,680		
M4.4	**Geotechnical Monitoring**												**$23,216**	
M4.4.1	Annual Geotech Inspection	Consultant travel/on-site time	2	ea	16.0	$50	$117		$100	$3,744		$3,844		
M4.4.2		Car Rental (2 days)	2	days				$350			$700	$700		
M4.4.3		Helicopter time	1	ea	5		$49	$1,600		$243	$8,000	$8,243		
M4.4.4		Instrumentation review	1	ls	24		$45			$1,070		$1,070		
M4.4.5		Reporting/Maintenance Planning	1	ls	80		$117			$9,360		$9,360		
M4.5	**Reclamation Monitoring**												**$10,530**	
M4.5.1		Consultant travel - Whitehorse to site	2	ls	8.1	$50	$117		$100	$1,892		$1,992		
M4.5.2		Truck Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M4.5.3		Inspect vegetation	2	ls	12		$117		$0	$2,808		$2,808		
M4.5.4		Reporting	1	ls	40		$117			$4,680		$4,680		
M4.6	**Annual Reclamation Reports**												**$23,900**	
M4.6.1		Reporting	1	ls	200	$500	$117		$500	$23,400		$23,900		Includes as-built reporting
TOTAL												**$100,548**	**$100,548**	

NOTES:
1. Water quality sampling assumes 2 rounds of sampling completed by helicopter from Carmacks, and one round when the barge is in place.
2. Assumes sediment samples collected during water quality sampling.
3. Assumes biological sampling completed by consultants based in Whitehorse, and sampling takes place while barge in place.
4. Geotechnical inspection assumed to be completed by helicopter. Two consultants assumed to be present to also collect geotech data. The inspection is also used to determine the site maintenance required for the year.
5. Assumes annual reclamation report compiled by outside consultants.

Phase 5 - Annual Monitoring Costs - Perpetual Maintenance Year

						Unit Rates			Activity Totals					
WBS	Facility/Area	Task	Crew/ Qnty	Unit	Hours	Material Cost ($/unit)	Labour Rate ($/hr)	Equipment Rate ($/hr)	Material Cost	Labour Cost	Equipment Cost	Cost	Sub-Totals	Source / Comments
M5.1	**Water Quality Monitoring**												**$27,664**	
M5.1.1		Helicopter time (3 rounds)	3	ea	2.25		$49	$1,600		$328	$10,800	$11,128		Helicopter cost based on onsite costs
M5.1.2		Consultant time incl. travel	3	ea	12		$117	$12		$4,212	$445	$4,657		1 consultant
M5.1.3		Car Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M5.1.4		Complete lab testing	41	ea		$150			$6,150			$6,150		
M5.1.5		Reporting	1	ea	40		$117			$4,680		$4,680		
M5.2	**Sediment Monitoring**												**$3,388**	
M5.2.1		Collect data/samples	2	ea	0		$45	$12		$0.00	$0.00	$0		Collected during water sampling
M5.2.2		Laboratory Analysis	2	ea		$290			$580			$580		
M5.2.3		Reporting	1	ls	24		$117			$2,808		$2,808		
M5.3	**Biological Monitoring**												**$12,294**	
M5.3.1		Consultant travel - Whitehorse to/from site	2	ls	8.1	$50	$117		$100	$1,892		$1,992		
M5.3.2		Truck Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M5.3.3		Collect data/samples	2	ls	8		$117			$1,872		$1,872		
M5.3.4		Laboratory Analysis	10	ea		$270			$2,700			$2,700		
M5.3.5		Reporting	1	ls	40		$117			$4,680		$4,680		
M5.4	**Geotechnical Monitoring**												**$23,216**	
M5.4.1	Annual Geotech Inspection	Consultant travel/on-site time	2	ea	16.0	$50	$117		$100	$3,744		$3,844		
M5.4.2		Car Rental (2 days)	2	days				$350			$700	$700		
M5.4.3		Helicopter time	1	ea	5		$49	$1,600		$243	$8,000	$8,243		
M5.4.4		Instrumentation review	1	ls	24		$45			$1,070		$1,070		
M5.4.5		Reporting/Maintenance Planning	1	ls	80		$117			$9,360		$9,360		
M5.5	**Reclamation Monitoring**												**$10,530**	
M5.5.1		Consultant travel - Whitehorse to site	2	ls	8.1	$50	$117		$100	$1,892		$1,992		
M5.5.2		Truck Rental (inc. fuel)	3	days				$350			$1,050	$1,050		
M5.5.3		Inspect vegetation	2	ls	12		$117			$2,808		$2,808		
M5.5.4		Reporting	1	ls	40		$117			$4,680		$4,680		
M5.6	**Annual Reclamation Reports**												**$23,900**	
M5.6.1		Reporting	1	ls	200	$500	$117		$500	$23,400		$23,900		Includes Asbuilt reporting
TOTAL												**$100,993**	**$100,993**	

NOTES:
1. Water quality sampling assumes 2 rounds of sampling completed by helicopter from Carmacks, and one round when the barge is in place.
2. Assumes sediment samples collected during water quality sampling.
3. Assumes biological sampling completed by consultants based in Whitehorse, and sampling takes place while barge in place.
4. Geotechnical inspection assumed to be completed by helicopter. Two consultants assumed to be present to also collect geotech data. The inspection is also used to determine the site maintenance required for the year.
5. Assumes annual reclamation report compiled by outside consultants and includes additional hours for asbuilt reporting

Worksheet 26 - Ore Stockpiles and Misc. Volume Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

References (dwgs/plans \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\!040_AutoCAD\1CM002-45_ClosureQuantities.dwg

A: Ore Stockpile Excavation

		Inputs				Calculated Quantities			
Area	Description	Volume of Ore Remaining (m^3)	Ore Pad Area (m^2)	Ore Pad Excavation depth (m)	Confirmation Test Grid Spacing (m)	Ore Relocation Volume (m^3)	Ore Pad Excavation Volume (m^3)	Confirmation Tests Required	Comments/Notes
Year 0 Scenario									
Mill Area	North Stockpile	70,673	13,760	0.5	10	70,673	6,880	138	
	West Stockpile	1,657	38,670	0.5	10	1,657	19,335	387	
	East Stockpile	0	31,190	0.5	10	0	15,595	312	
	South Stockpile	123,698	41,571	0.5	10	123,698	20,786	416	
	Nuway Stockpiles	9,463	4,145	0.5	10	9,463	2,073	41	
	Crusher Stockpile	0	12,678	0.5	10	0	6,339	127	
Mill Area	**TOTAL**					205,490	71,007	1,421	
Main Waste Dump	Low Grade Ore Stockpiles	15,490	62,631	0.5	10	15,490	31,316	626	"Top of the World Stockpile"
Main Waste Dump	**TOTAL**					15,490	31,316	626	
SWD	High grade waste	45,711				45,711			
South West Dump	**TOTAL**					45,711	0	0	
Peak Liability Scenario									
Mill Area	North Stockpile	0	13,760	0.5	10	0	6,880	138	
	West Stockpile	0	38,670	0.5	10	0	19,335	387	
	East Stockpile	0	31,190	0.5	10	0	15,595	312	
	South Stockpile	884,785	41,571	0.5	10	884,785	20,786	416	All mill area ore was allocated to this stockpile.
	Crushed Ore Stockpiles	0	4,145	0.5	10	0	2,073	41	
	Crusher Stockpile	0	12,678	0.5	10	0	6,339	127	
Mill Area	**TOTAL**					884,785	71,007	1,421	
Main Waste Dump	Low Grade Ore Stockpiles	15,490	62,631	0.5	10	15,490	31,316	626	
Main Waste Dump	**TOTAL**					15,490	31,316	626	
SWD	High grade waste	44,730				44,730			Based of HGW covered as per SWD perscription
South West Dump	**TOTAL**					44,730	0	0	

Notes:
1. Year 0 stockpile tonnes provided by Minto in January 2018. Stockpile density = 2 tonnes/m3
2. Peak ore stockpile volume occurs in 2019 and is estiamted based on the closure plan production profile. For the EOM scenario, the stockpile volumes are 0.
3. Covers and revegetation of the ore stockpiles included in the "Areas" worksheet.

B: Misc. Volumes

		Inputs				Calculated Quantities			
Area	Description	Length (m)	Width (m)	Area (m2)	Average Thickness (m)	Volume (Cm3)	Volume (Lm3)		Comments/Notes
DSTSF	Fill at south side of DSTSF to cover tailings and provide drainage to the east.	404	25		3	30,300			Material assumed to be sourced from the Tailings Diversion Ditch road
DSTSF	Unsuitable existing cover material to be removed.			69,467	0.75	52,100			Area based on 2014 aerial photo
Main Pit/SAT	SAT in the Main Pit to be relocated to below 786 m elevation					1,040,000			Volume provided by Minto Dec. 2015, and includes a projection of the SAT volume to be placed from Minto North
Main Pit/SAT	Allowance for backfill to stabilize MPD/Main Pit South Wall following removal of SAT					245,000			Buttress volume from E. Ketilison from AutoCAD
Main Pit	Est. Tailings near outlet above long term water table elevation of 786.			2,311	1.50	3,467			Area estimated from 2014 site aerial photographs and June 2016 site photography
Southwest Wetlands	3 Berms acrros valley	300	5		2.00	3,000			Assumed width, thickness. Length estimated from 2014 orthophoto.
Ridgetop North	Waste Rock Volume						3,416,000	0.84	From WROMP June 2014
Ridgetop South	Waste rock volume						637,000	0.16	From WROMP June 2014

Notes:

Worksheet 27 - Open Pit Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

References (dwgs/plans): \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\!040_AutoCAD\1CM002-45_ClosureQuantities.dwg
Layer: ~Pits

Open Pit Quantity Calculations

		Safety Berms							Boulder Fence		Warning Signs			Quantity Calculations					
Pit	Scenario	Berm Placement Length (m)	% Open Space in Berm	Length of berm requiring land clearing (m)	Land clearing width (m)	Berm Height (m)	Berm Side-Slope Angle (H:1V)	Berm Crest Width (m)	Fence Length (m)	Boulder Spacing (m)	Warning Sign Perimeter Length (m)	Sign Spacing (m)	User Override - Number of Signs	Berm Volume (m^3/m)	Berm Volume (m^3)	Land Clearing Area (m^2)	Boulder Fence Volume (m^3)	Warning Signs	Comments/Notes
Main Pit	All	1680	14%	750	10	1.5	1.3	0.3	40	1	2741	100	6	3.4	4,860	7,500	52	6	
Area 2 Pit (updated for 2017	Yr0	1838	14%	0	10	1.5	1.3	0.3	20	1	1506	100	6	3.4	5,317	0	26	6	
Area 2 Pit (inc. Stage 3)	EOM	1838	14%	0	10	1.5	1.3	0.3	20	1	1506	100	6	3.4	5,317	0	26	6	
Area 118 Pit	All	496	14%	258	10	1.5	1.3	0.3	20	1	600	100	4	3.4	1,435	2,580	26	4	
Minto North	All	1248	14%	590	10	1.5	1.3	0.3	20	1	1248	100	4	3.4	3,610	5,900	26	4	
Ridgetop North	Peak	1261	14%	1011	10	1.5	1.3	0.3	20	1	1506	100	6	3.4	3,648	10,110	26	6	
Ridgetop South	Peak	534	14%	534	10	1.5	1.3	0.3	20	1	1506	100	6	3.4	1,545	5,340	26	6	

Notes:
1. Vegetated areas of the highwall require clearing to ensure visibility of the safety berm.
2. Main Pit Rock fence assumed to be placed near the entrance ramp of the pit.

Worksheet 28 - Resloping Quantity Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

Notes/Instructions: 1. When adding a new row, make sure all columns are copied (there are a lot of calculations hidden in columns to the right).
2. Once complete, check that the Dozer used on this sheet matches the correct task unit rate used on the cost estimate worksheet
References (dwgs/plans): \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\l040_AutoCAD\1CM002-45_ClosureQuantities.dwg
(Layer ~Regrading)

A: Slopes

		Dump							Dozing Conditions					Calculated Quantities					Total Quantities						
Facility	Segment	Mid-bench length (m)	Height (m)	Existing Side Slope Angle (H:1V)	Final Side-Slope Angle (H:1V)	Underlying slope grade (%) (See Note 2)	Does regrading increase the dump footprint?	Soil Cover Thickness (m)	Dozer Size	Material Type	Dozing Condition	Bank or Placed? (See note 3)	Slot or Side-by-side Dozing?	Cut Volume per m crest (m^3)	Average Regrade Push Distance (m)	Unfactored Dozer Productivity (Lm^3/hr)	Factored Dozer Productivity (Bm^3/hr)	Time to complete 1m width (hrs)	Time Required - Regrading (Hrs)	Time Required - Spreading Cover (hrs)	Total Regrade Volume (Bm^3)	3D Surface Area (m^2)	2D Sloped Area (m^2)	2D Footprint increase due to regrading (m^2)	Comments/Notes
Existing Facilities																									
Camp Area	R1	309	18	1.3	3	0	Yes	0.5	D10	Sand & gravel,	Normal	Placed	No	68.9	37.9	1114.1	873.5	0.1	24	10	21,275	17,589	16,686	4,728	
Camp Area	R2	73	3	1.3	3	0	No	0.5	D10	Sand & gravel,	Normal	Placed	No	1.9	6.3	2293.7	1798.4	0.0	0	0	140	693	657	0	
Camp Area	R3	44	3	1.3	3	0	Yes	0.5	D10	Sand & gravel,	Normal	Placed	No	1.9	6.3	2293.7	1798.4	0.0	0	0	84	417	396	112	
Camp Area	**TOTAL**																		**24**	**10**	**21,498**	**18,699**	**17,739**	**4,840**	
DSTSF - WR Shell	R1	3211	4	1.3	4	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	5.4	11.0	2293.7	1309.0	0.0	13	14	17,339	52,957	51,376	0	
DSTSF - WR Shell	**TOTAL**																		**13**	**35**	**17,339**	**52,957**	**51,376**	**0**	
DSTSF - South end	R1	500	5	1.3	4	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	8.4	13.7	2293.7	1309.0	0.0	3	3	4,219	10,308	10,000	0	
DSTSF - South End	**TOTAL**																		**3**	**72**	**4,219**	**10,308**	**10,000**	**0**	
IROD Laydown	R1	101	14	3.5	3.5	0	No	0	D10	Sand & gravel,	Normal	Placed	No	0.0	0.0	2293.7	1787.2	0.0	0	0	0	5,147	4,949	0	
IROD Laydown	**TOTAL**																		**0**	**0**	**0**	**5,147**	**4,949**	**0**	
Main Dump (2014)	R1	282	60	2.5	3	2	Yes	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	226.2	127.2	377.5	227.7	1.0	280	89	63,786	53,790	51,029	4,489	
Main Dump	R2	185	35	2.5	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	76.6	73.8	614.4	370.7	0.2	38	21	14,164	20,476	19,425	0	
Main Dump	R3	358	15	1.3	3	0	Yes	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	47.8	31.6	1311.6	791.2	0.1	22	8	17,117	16,981	16,110	4,565	
Main Dump	R5	266	22	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	102.9	46.4	931.0	561.6	0.2	49	12	27,358	18,506	17,556	0	
Main Dump	R6	73	17	1.5	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	54.2	35.8	1172.6	707.4	0.1	6	2	3,956	3,924	3,723	0	
Main Dump	R7	50	7	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	10.4	14.8	2293.7	1383.7	0.0	0	0	521	1,107	1,050	0	
Main Dump	R11	200	20	1.3	3	-75	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	65.2	32.9	1267.0	764.4	0.1	17	5	13,049	9,861	9,355	0	
Main Dump	R12	75	5	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	5.3	10.5	2293.7	1383.7	0.0	0	0	398	1,186	1,125	0	
Main Dump	R13	93	20	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	85.0	42.2	1013.9	611.6	0.1	13	4	7,905	5,882	5,580	0	
Main Dump	R14	123	30	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	191.3	63.2	705.3	425.5	0.4	55	10	23,524	11,669	11,070	0	
Main Dump	R15	205	10	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	21.3	21.1	1885.4	1137.4	0.0	4	2	4,356	6,483	6,150	0	
Main Dump	R16	112	20	1.3	3	-75	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	65.2	32.9	1267.0	764.4	0.1	10	3	7,307	5,522	5,239	0	
Main Dump Extension	R17	964	15	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	47.8	31.6	1311.6	791.2	0.1	58	22	46,091	45,727	43,380	0	
Main Dump Extension	R18-	990	15	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	47.8	31.6	1311.6	791.2	0.1	60	22	47,334	46,960	44,550	0	
Main Dump Extension	R19	1000	15	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	47.8	31.6	1311.6	791.2	0.1	60	23	47,813	47,434	45,000	0	
MWD & MWDE (Year 0)	**TOTAL**																		**672**	**224**	**324,679**	**295,506**	**280,341**	**9,053**	
Main Dump Wrap	R1	560	20	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	85.0	42.2	1013.9	611.6	0.1	78	22	47,600	35,418	33,600	0	
Main Dump Wrap	R2	600	20	1.3	4	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	135.0	55.0	799.6	456.3	0.3	178	39	81,000	49,477	48,000	0	
Main Dump Wrap	**TOTAL**																		**255**	**61**	**128,600**	**84,895**	**81,600**	**0**	
Main Pit Dump - Yr0	R1	277	16	1.3	3	15	Yes	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	65.2	40.8	1043.9	629.8	0.1	29	10	18,066	16,956	16,086	6,268	
Main Pit Dump - Yr0	R2	566	15	1.3	3	0	Yes	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	47.8	31.6	1311.6	791.2	0.1	34	13	27,062	26,848	25,470	7,217	
Main Pit Dump - Yr0	R3	384	15	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	47.8	31.6	1311.6	791.2	0.1	23	9	18,360	18,215	17,280	0	
Main Pit Dump - Yr0	R4	711	15	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	47.8	31.6	1311.6	791.2	0.1	43	16	33,995	33,726	31,995	0	
Main Pit Dump - Yr0	R5	700	12	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	30.6	25.3	1601.5	966.2	0.0	22	10	21,420	26,563	25,200	0	
Main Pit Dump - Yr0	R6	100	21	1.5	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	82.7	44.3	970.5	585.5	0.1	14	4	8,269	6,641	6,300	0	
Main Pit Dump - Yr0	R7	226	14	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	41.7	29.5	1395.1	841.6	0.0	11	4	9,413	10,005	9,492	0	
Main Pit Dump - Yr0	R8	354	11	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	25.7	23.2	1731.2	1044.4	0.0	9	4	9,102	12,314	11,682	0	
Main Pit Dump - Yr0	R9	355	15	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	47.8	31.6	1311.6	791.2	0.1	21	8	16,973	16,839	15,975	0	
Main Pit Dump - Yr 0 (2018	**TOTAL**																		**207**	**79**	**162,660**	**168,106**	**159,480**	**13,485**	
MVFE Stage 1 and 2	R1	255	17	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	61.4	35.8	1172.6	707.4	0.1	22	7	15,660	13,708	13,005	0	
MVFE Stage 1 and 2	R2	188	12	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	30.6	25.3	1601.5	966.2	0.0	6	3	5,753	7,134	6,768	0	
MVFE Stage 1 and 2	R3	250	10	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	21.3	21.1	1885.4	1137.4	0.0	5	3	5,313	7,906	7,500	0	
MVFE Stage 1 and 2	R4	265	12	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	30.6	25.3	1601.5	966.2	0.0	8	4	8,109	10,056	9,540	0	
MVFE Stage 1 and 2	R5	400	5	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	5.3	10.5	2293.7	1383.7	0.0	2	2	2,125	6,325	6,000	0	
MVFE Stage 1 and 2	**TOTAL**																		**43**	**18**	**36,959**	**45,129**	**42,813**	**0**	
Minto South Portal	R1	200	7	1.3	3	0	Yes	0	D10	Waste Rock (M	Rock, ripped or	Placed	No	10.4	14.8	2293.7	1383.7	0.0	2	0	2,083	4,427	4,200	1,190	
Minto South Portal	R2	133	10	1.3	3	10	Yes	0	D10	Waste Rock (M	Rock, ripped or	Placed	No	23.6	23.5	1710.6	1032.0	0.0	3	0	3,142	4,689	4,448	1,535	
Minto South Portal	R3	200	7	1.3	3	10	No	0	D10	Waste Rock (M	Rock, ripped or	Placed	No	11.6	16.5	2353.9	1420.0	0.0	2	0	2,315	4,936	4,682	0	
Minto South Portal	R4	80	10	1.3	3	10	No	0	D10	Waste Rock (M	Rock, ripped or	Placed	No	23.6	23.5	1710.6	1032.0	0.0	2	0	1,890	2,820	2,676	0	
Minto South Portal	R5	490	20	2.5	3	0	No	0	D10	Waste Rock (M	Rock, ripped or	Placed	No	25.0	42.2	1013.9	611.6	0.0	20	0	12,250	30,990	29,400	0	
Minto South Portal	**TOTAL**																		**28**	**0**	**21,678**	**47,862**	**45,406**	**2,725**	
Southwest Dump	Allowance	1000	15	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	47.8	31.6	1311.6	791.2	0.1	60	23	47,813	47,434	45,000	0	
Southwest Dump	**TOTAL**																		**60**	**23**	**47,813**	**47,434**	**45,000**	**0**	
SWD - High Grade Waste	R1	320	8	1.3	5	0	Yes	1	D10	Waste Rock (M	Rock, ripped or	Placed	No	29.6	27.2	1501.2	796.2	0.0	12	11	9,472	13,053	12,800	4,736	
SWD - High Grade Waste	R2	113	11	1.3	5	0	Yes	1	D10	Waste Rock (M	Rock, ripped or	Placed	No	56.0	37.4	1128.9	598.7	0.1	11	7	6,324	6,338	6,215	2,300	
SWD - High Grade Waste	R3	245	17	1.3	5	2	Yes	1	D10	Waste Rock (M	Rock, ripped or	Placed	No	139.0	60.1	738.1	391.4	0.4	87	37	34,054	22,093	21,664	8,394	
SWD - High Grade Waste	**TOTAL**																		**109**	**55**	**49,849**	**41,485**	**40,679**	**15,430**	
Tailings Diversion Ditch	R1	260	7	1.3	3	17	Yes	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	13.0	18.6	2108.3	1271.9	0.0	3	2	3,368	7,257	6,884	2,833	
Tailings Diversion Ditch	R2	454	13	1.3	3	10	Yes	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	39.9	30.6	1352.6	816.0	0.0	22	10	18,123	20,807	19,739	6,811	
Tailings Diversion Ditch	R3	311	7	1.3	3	13	Yes	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	12.1	17.2	2259.2	1362.9	0.0	3	2	3,757	8,035	7,623	2,823	
Tailings Diversion Ditch	**TOTAL**																		**28**	**14**	**25,248**	**36,099**	**34,246**	**12,467**	
W15 Sump Area	R1	124	6	1.3	3	0	Yes	0	D10	Waste Rock (M	Rock, ripped or	Placed	No	7.7	12.6	2293.7	1383.7	0.0	1	0	949	2,353	2,232	632	
W15 Sump Area	R2	92	6	1.3	3	15	Yes	0	D10	Waste Rock (M	Rock, ripped or	Placed	No	9.2	15.3	2511.4	1515.1	0.0	1	0	844	2,112	2,003	781	
Minto South Portal	**TOTAL**																		**1**	**0**	**1,792**	**4,465**	**4,235**	**1,413**	
Future Facilities																									
Area 118 Backfill Dump	R1	1227	5	1.3	3	0	No	0	D10	Earth, moist	Normal	Placed	No	5.3	10.5	2293.7	1648.3	0.0	4	0	6,518	19,401	18,405	0	Source: A118 Backfill Dump.dwg
Area 118 Backfill Dump	**TOTAL**																		**4**	**0**	**6,518**	**19,401**	**18,405**	**0**	
Main Pit Dump - EOM	R1-EOM	1284	15	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	47.8	31.6	1311.6	791.2	0.1	78	29	61,391	60,905	57,780	0	
Main Pit Dump - EOM	R6	203	21	1.5	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	82.7	44.3	970.5	585.5	0.1	29	9	16,786	13,481	12,789	0	
Main Pit Dump - EOM	R7	176	14	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	41.7	29.5	1395.1	841.6	0.0	9	3	7,330	7,792	7,392	0	
Main Pit Dump - EOM	R8	170	11	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	25.7	23.2	1731.2	1044.4	0.0	4	2	4,371	5,913	5,610	0	
Main Pit Dump - EOM	R9	355	15	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	47.8	31.6	1311.6	791.2	0.1	21	8	16,973	16,839	15,975	0	
Main Pit Dump - EOM	**TOTAL**																		**141**	**51**	**106,852**	**104,931**	**99,546**	**0**	
Ridgetop South Backfill Dum	R1	257	15	1.3	3	17	Yes	0	D10	Waste Rock (M	Rock, ripped or	Placed	No	59.5	39.9	1065.8	643.0	0.1	24	0	15,287	15,371	14,582	6,001	
Ridgetop South Backfill Dum	R2	253	10	1.3	3	0	No	0	D10	Waste Rock (M	Rock, ripped or	Placed	No	21.3	21.1	1885.4	1137.4	0.0	5	0	5,376	8,001	7,590	0	
Ridgetop South Backfill Dum	R3	200	10	1.3	3	0	No	0	D10	Waste Rock (M	Rock, ripped or	Placed	No	21.3	21.1	1885.4	1137.4	0.0	4	0	4,250	6,325	6,000	0	
Ridgetop Waste Dump	**TOTAL**																		**32**	**0**	**24,913**	**29,696**	**28,172**	**6,001**	
Ridgetop Waste Dump	R1	919	20	1.3	3	17	Yes	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	105.7	53.2	823.9	497.0	0.2	196	56	97,182	73,285	69,524	28,609	
Ridgetop Waste Dump	R2	1183	15	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	47.8	31.6	1311.6	791.2	0.1	71	27	56,562	56,115	53,235	0	
Ridgetop Waste Dump	R3	1495	10	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	21.3	21.1	1885.4	1137.4	0.0	28	16	31,769	47,276	44,850	0	
Ridgetop Waste Dump	R4	1265	10	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	21.3	21.1	1885.4	1137.4	0.0	24	13	26,881	40,003	37,950	0	
Ridgetop Waste Dump	R5	203	10	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	21.3	21.1	1885.4	1137.4	0.0	4	2	4,314	6,419	6,090	0	
Ridgetop Waste Dump	R6	1530	13	1.3	3	0	No	0.5	D10	Waste Rock (M	Rock, ripped or	Placed	No	35.9	27.4	1490.8	899.4	0.0	61	26	54,946	62,898	59,670	0	
Ridgetop Waste Dump	**TOTAL**																		**383**	**140**	**271,654**	**285,996**	**271,319**	**28,609**	

Notes:
1. Regrading productivity calculations/details are provided in below (spreadsheet).
2. A positive grade indicates the toe is being extended downward; a negative valley indicates the dump toe is moving uphill.
2. Bank materials are insitu, natural soils. Placed are materials that have been placed and compacted and are generally denser than Bank/In-situ materials. See densities and bulking factors below (spreadsheet). **For Waste Rock - Use 'Placed"**.

B: Flat Area Tertiary Catchments

	Dimensions				Dozing Conditions				Calculated Quantities							Total Quantities		
Facility	Total Flat Area (m^2)	Maximum Tertiary Catchment Area (m^2)	Avg. Final Slope (%)	Length of dozing direction over-ride (m)	Dozer Size	Material Type	Dozing Condition	Bank or Placed? (See note 3)	Number of catchments	Catchment Length (in dozing direction) (m)	Average Regrade Push Distance (m)	Unfactored Dozer Productivity (Lm^3/hr)	Factored Dozer Productivity (Bm^3/hr)	Cut Volume Per Catchment (Bm^3)	Time to complete 1 catchment (hrs)	**Total Time Required (Hrs)**	**Avg. Regrade Productivity (m^2/hr)**	**Comments/Notes**
A118 Backfill dump (2018)	41,899	15,000	3%	50	D10	Sand & gravel	Normal	Placed	3	50.0	33.3	1251.2	560.6	2,813	5.0	14	2,990	
A118 Backfill dump (EOM)	50,033	15,000	3%	50	D10	Sand & gravel	Normal	Placed	3	50.0	33.3	1251.2	560.6	2,813	5.0	17	2,990	
Camp Area	20,626	15,000	3%	50	D10	Sand & gravel	Normal	Placed	1	50.0	33.3	1251.2	560.6	2,813	5.0	7	2,990	
Crusher Area	26,964	15,000	3%	50	D10	Sand & gravel	Normal	Placed	2	50.0	33.3	1251.2	560.6	2,813	5.0	9	2,990	
DSTSF	0	15,000	1%		D10	Earth, moist	Normal	Placed	0	122.5	81.6	561.2	241.3	2,296	9.5	0	#DIV/0!	
Explosive Plant/Storage area	55,113	15,000	3%	60	D10	Waste Rock (	Hard to cut; fro	Placed	4	60.0	40.0	1062.8	418.8	3,375	8.1	30	1,861	
Fuel Storage Area	8,400	15,000	3%	60	D10	Waste Rock (	Hard to cut; fro	Placed	1	60.0	40.0	1062.8	418.8	3,375	8.1	5	1,861	
Ice Rich Overburden Dump	0	15,000	3%	80	D10	Earth, moist	Normal	Placed	0	80.0	53.3	821.5	337.4	4,500	13.3	0	#DIV/0!	
IROD Laydown	18,106	15,000	3%	80	D10	Sand & gravel	Normal	Placed	1	80.0	53.3	821.5	368.1	4,500	12.2	15	1,227	
Main Dump	90,055	15,000	3%		D10	Waste Rock (	Hard to cut; fro	Placed	6	122.5	81.6	561.2	221.1	6,889	31.2	187	481	Total Flat Area was reduced by 13.5 ha (flat area at the top of the MWDE) that is being constructed at a 2% grade during ops.
Main Dump Wrap	16,446	15,000	3%	50	D10	Waste Rock (	Hard to cut; fro	Placed	1	50.0	33.3	1251.2	493.0	2,813	5.7	6	2,629	
Main Pit Dump (Yr 0)	90,540	15,000	3%	70	D10	Waste Rock (	Hard to cut; fro	Placed	6	70.0	46.7	925.8	364.8	3,938	10.8	65	1,390	
Main Pit Dump (EOM)	#REF!	15,000	3%	70	D10	Waste Rock (	Hard to cut; fro	Placed	#REF!	70.0	46.7	925.8	364.8	3,938	10.8	#REF!	#REF!	
Mill Area	42,365	15,000	3%	100	D10	Waste Rock (	Hard to cut; fro	Placed	3	100.0	66.7	672.8	265.1	5,625	21.2	60	707	
MVFE Stage 1 and 2	114,853	15,000	3%	100	D10	Waste Rock (	Hard to cut; fro	Placed	8	100.0	66.7	672.8	265.1	5,625	21.2	162	707	
Minto South Portal	62,765	15,000	3%	75	D10	Waste Rock (	Hard to cut; fro	Placed	4	75.0	50.0	870.4	343.0	4,219	12.3	51	1,219	
Ore Stockpile Area	139,307	15,000	3%		D10	Waste Rock (	Hard to cut; fro	Placed	9	122.5	81.6	561.2	221.1	6,889	31.2	289	481	
Pelly Laydown	51,710	15,000	3%		D10	Waste Rock (	Hard to cut; fro	Placed	3	122.5	81.6	561.2	221.1	6,889	31.2	107	481	
Reclamation OVB Dump	297,857	15,000	3%		D10	Earth, moist	Normal	Placed	20	122.5	81.6	561.2	230.5	6,889	29.9	594	502	
Ridgetop South Backfill Dum	8,491	15,000	3%		D10	Earth, moist	Normal	Placed	1	122.5	81.6	561.2	230.5	6,889	29.9	17	502	
Ridgetop Waste Dump	44,126	15,000	3%		D10	Earth, moist	Normal	Placed	3	122.5	81.6	561.2	230.5	6,889	29.9	88	502	
SWD - High Grade Waste	15,583	15,000	3%		D10	Earth, moist	Normal	Placed	1	122.5	81.6	561.2	230.5	6,889	29.9	31	502	
W15 Sump Area	9,484	15,000	3%	80	D10	Waste Rock (	Hard to cut; fro	Placed	1	80.0	53.3	821.5	323.7	4,500	13.9	9	1,079	

Notes:
1. Regrading productivity calculations/details are provided in below (spreadsheet).
2. Productivity calculations assumes the dozer is pushing uphill
3. Dozer pushing distance calculations assume the tertiary catchments are approximately square, unless the 'length of dozing direction' user over-ride is used.
4. Bank materials are insitu, natural soils. Placed are materials that have been placed and compacted and are generally denser than Bank/In-situ materials (See densities and bulking factors below). **For Waste Rock - Use 'Placed"**.

C: Miscellaneous Regrading

Facility	Volume to Push (m^3)	Avg. Grade (%)	Dozer Size	Material Type	Dozing Condition	Bank or Placed? (See note 3)	Average Regrade Push Distance (m)	Unfactored Dozer Productivity (Lm^3/hr)	Factored Dozer Productivity (Bm^3/hr)	**Total Time Required (Hrs)**	Comments/Notes
Mill Pond Backfilling	7,088	0%	D10	Waste Rock (	Hard to cut, fr	Placed	50.0	870.4	367.0	**19**	Assumes material pushed from the north (crusher area)
SWD Cover placement on st	0	20%	D10	Earth, moist	Loose Stockp	Placed	40.0	1062.8	315.4	**0**	

Notes:
1. Regrading productivity calculations/details are provided in below (spreadsheet).
2. Productivity calculations assumes the dozer is pushing uphill
3. Dozer pushing distance calculations assume the tertiary catchments are approximately square, unless the 'length of dozing direction' user over-ride is used.
4. Bank materials are insitu, natural soils. Placed are materials that have been placed and compacted and are generally denser than Bank/In-situ materials (See densities and bulking factors below). **For Waste Rock - Use 'Placed"**.

Worksheet 29 - Revegetation Prescriptions

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

A: Tractor-Applied Seed Mixes

Seed Mix 1 - DRY AREA SEED MIX

Item	Target Application Rate (kg/ha)	Over-application Allowance (%)	Applied Rate (kg/ha)	Unit Cost ($/kg)	Item Cost ($/ha)	Comment
Seed Mix	30		45	$16.12	$725.22	
Fertilizer	200	25%	250	$1.12	$280.50	
Mulch	0	0%	0	$1.02		
Tackifier	0	0%	0	$3.69		
TOTAL SEED MIX COST PER HA:					**$1,005.72**	

Seed Mix 2 - WET AREA SEED MIX

Item	Target Application Rate (kg/ha)	Over-application Allowance (%)	Applied Rate (kg/ha)	Unit Cost ($/kg)	Item Cost ($/ha)	Comment
Seed Mix	30		45	$16.12	$725.22	
Fertilizer	200	25%	250	$1.12	$280.50	
Mulch	0	0%	0	$1.02	$0.00	
Tackifier	0	0%	0	$3.69	$0.00	
TOTAL SEED MIX COST PER HA:					**$1,005.72**	

B: Tree Planting

Item	Target Application Rate (stem/ha)	Over-application Allowance (%)	Applied Rate (kg/ha)	Unit Cost ($/kg)	Item Cost ($/ha)	Comment
Trees	Seedlings per ha					
Tree seedlings	1,000	0%	1,000			Target application rates from Closure Plan
Subtotal - Seed Mix	1000		1000	$0.39	$388	
Fertilizer	1000	0%	1000	$0.08	$82.11	
TOTAL TREE PLANTING COST PER HA:					**$470**	

Worksheet 30-Road Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

References (dv\\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\\040_AutoCAD\1CM002-45_ClosureQuantities.dwg

A: Exploration Roads

		Road Parameters						Safety Berms				Reclamation Prescription				X-Section Quantities				Total Quantities				
Area	Description	Length (m)	Width (at crest) (m)	Average Fill Height (m)	Number of slopes to regrade (0, 1 or 2)	Existing Side Slope Angle (H:1V)	Final Side-Slope Angle (H:1V)	Number of Berms (0, 1 or 2)	Berm Height (m)	Berm Side-Slope Angle (H:1V)	Berm Crest Width (m)	Dozer Size	Scarify (y/n)	Growth Media Thickness (m)	Revegetate (y/n)	Berm Volume (1-side) (m^3/m)	Regrade Cut Volume (m^3/m)	Average Regrade Push Distance (m)	Factored Dozer Productivity (m/hr)	Scarification Area (m^2)	Cover Volume (Cm^3)	Revegetation Area (ha)	Regrade Hours	Comments/Notes
Airstrip East	Area east of airstrip	4581	3		0	1	3	0	0	0	0		Yes	0	Yes	0.0	0.0	0.0	#N/A	13,743	0	1.37	0.0	
Airstrip West	Area west of airstrip	4299	3		0	1	3	0	0	0	0		Yes	0	Yes	0.0	0.0	0.0	#N/A	12,897	0	1.29	0.0	
Ridgetop		4252	3		0	1	3	0	0	0	0		Yes	0	Yes	0.0	0.0	0.0	#N/A	12,756	0	1.28	0.0	
North	North of Main Pit	4963	3		0	1	3	0	0	0	0		Yes	0	Yes	0.0	0.0	0.0	#N/A	14,889	0	1.49	0.0	
TOTALS																				**54,285**	**0**	**5**	**0**	

Notes:
1. Dozer regrading productivity calculations details are provided in the "reslope" worksheet.
2. If a side slope is not regraded, it is assumed that it is not revegetated, and no cover is placed.

B: Access Roads

		Road Parameters						Safety Berms				Reclamation Prescription				X-Section Quantities				Total Quantities				
Area	Description	Length (m)	Width (at crest) (m)	Average Fill Height (m)	Number of slopes to regrade (0, 1 or 2)	Existing Side Slope Angle (H:1V)	Final Side-Slope Angle (H:1V)	Number of Berms (0, 1 or 2)	Berm Height (m)	Berm Side-Slope Angle (H:1V)	Berm Crest Width (m)	Dozer Size	Scarify (y/n)	Growth Media Thickness (m)	Revegetate (y/n)	Berm Volume (1-side) (m^3/m)	Regrade Cut Volume (m^3/m)	Average Regrade Push Distance (m)	Factored Dozer Productivity (m/hr)	Scarification Area (m^2)	Cover Volume (Cm^3)	Revegetation Area (ha)	Regrade Hours	Comments/Notes
Airstrip	Airport Access Road	968	14	1.5	2	1.3	3	2	1	1.3	0	D8	Yes	0	Yes	1.3	3.2	3.2	266.0	13,552	0	2.27	3.6	
	Betw. Landfill/waste storage	236	10	1.5	0	1.3	3	1	1	1.3	0	D8	Yes	0	Yes	1.3	1.3	3.2	648.8	2,360	0	0.24	0.4	
	Old Exploration camp road	460	10	1.5	0	1.3	3	1	1	1.3	0	D8	Yes	0	Yes	1.3	1.3	3.2	648.8	4,600	0	0.46	0.7	
Camp	2 Access Roads	238	14	1.5	2	1.3	3	2	1	1.3	0	D8	Yes	0	Yes	1.3	3.2	3.2	266.0	3,332	0	0.56	0.9	
Explosives	Roads between explosive plant	1633	10	2	1	1.3	3	1	1	1.3	0	D8	Yes	0	Yes	1.3	1.7	4.2	501.9	16,330	0	2.67	3.3	
IROD Laydown	Road between yard areas	703	10	0	0	1.3	3	0	1	1.3	0	D8	Yes	0	Yes	1.3	0.0	0.0	#DIV/0!	7,030	0	0.70	0.0	
W15 Laydown	2 "F" shaped roads	388	14	0	1	1.3	3	1	1	1.3	0	D8	Yes	0	Yes	1.3	1.3	0.0	648.8	5,432	0	0.54	0.6	
TOTALS																				**52,636**	**0**	**7**	**9**	

Notes:
1. Dozer regrading productivity calculations details are provided in the "reslope" worksheet.
2. If a side slope is not regraded, it is assumed that it is not revegetated, and no cover is placed.

C: Haul Roads

		Road Parameters						Safety Berms				Reclamation Prescription				X-Section Quantities				Total Quantities				
Area	Description	Length (m)	Width (at crest) (m)	Average Fill Height (m)	Number of slopes to regrade (0, 1 or 2)	Existing Side Slope Angle (H:1V)	Final Side-Slope Angle (H:1V)	Number of Berms (0, 1 or 2)	Berm Height (m)	Berm Side-Slope Angle (H:1V)	Berm Crest Width (m)	Dozer Size	Scarify (y/n)	Growth Media Thickness (m)	Revegetate (y/n)	Berm Volume (1-side) (m^3/m)	Regrade Cut Volume (m^3/m)	Average Regrade Push Distance (m)	Factored Dozer Productivity (m/hr)	Scarification Area (m^2)	Cover Volume (Cm^3)	Revegetation Area (ha)	Regrade Hours	Comments/Notes
Main Pit	Mill to Main Dump	1193	28	1.5	0	1.3	3	1	1.5	1.3	0	D8	Y	0	Yes	2.9	2.9	3.2	288.3	33,404	0	3.34	4.1	
Area 118	Area 118 Backfill Road	1060	28	1.5	2	1.3	3	2	1.5	1.3	0	D8	Y	0	Yes	2.9	6.7	3.2	125.8	29,680	0	3.97	8.4	
Minto North Hai	MWDE to Minto North	1000	28	1.5	2	1.3	3	2	1.5	1.3	0	D8	Y	0	Yes	2.9	6.7	3.2	125.8	28,000	0	3.75	8.0	
TDD Road	Area costed as WR dump															0.0				0	0	0.00	0.0	Area costed as WR Dump
TOTALS																				**91,084**	**0**	**11**	**21**	

Notes:
1. Dozer regrading productivity calculations details are provided in the "reslope" worksheet.
2. If a side slope is not regraded, it is assumed that it is not revegetated, and no cover is placed.
3. Most haul road costs are included within their respective areas.

D: Culvert Removal

		Input Parameters							Total Quantities			
Area	Description	Qnty	Culvert length (m)	Culvert Diameter (m)	Average Excavation Depth (m)	Excavation Side-Slope (H:1V)	Excavation Base Width (m)	Mobilization time (hrs)	Mobilization/Demobilization Hours	Excavation Volume (BCM)	Demolition Volume (LCM)	Comments/Notes
Mill Water Pond	Upstream Culvert	1	210	1.4	4	2	2	0	1.0	8,400	69	Average excavation depth assumed
	Tailings line to Main Pit	1	38	1	4	2	2	0	0.0	1,520	9	diameter assumed
	Reclaim line culvert by ore stockpile	1	35	1	2	2	2	0	0.0	420	8	diameter assumed
Mill	Tailings line culverts between confluence area and mill	4	50	1	2	2	2	0	0.0	2,400	12	diameter assumed
	Culverts between mill and main access road	2	50	1	2	2	2	0	0.0	1,200	12	diameter assumed
Area 2	Tailings discharge culvert by Area 2 pit	**1**	86	1	2	2	2	0	1.0	1,032	20	diameter assumed
	Culvert for water line between Main and Area 2 pit	**1**	54	1	2	2	2	0	0.0	648	13	diameter assumed
Pelly laydown		**1**	200	1	2	2	2	0	1.0	2,400	47	diameter assumed
TOTAL									**3.0**	**18020.0**	**190.1**	

Notes:
1. Demolition volume assumes culvert is flattened and then has an average height of 15 cm.

Worksheet 31 - Surface Infrastructure Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

References (dwgs/plans): \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\!040_AutoCAD\1CM002-45_ClosureQuantities.dwg
Layer: ~Pipelines

A. Conveyance Pipelines

Calculation Inputs
Pipe Debris Bulking Factor: 1.30

					Pipeline Details					Calculated Quantities				
										HDPE Pipe Demolition Lengths (m)				
Line	Type	Details	From	To	Length (m)	Pipe Diameter (inches)	Total Diameter (incl. insulation) (inch)	Pipe Diameter Category	Lines to flush	0.75 to 4 inch	6 to 8 inches	10 to 18 inches	20 to 36 inches	Pipe Debris Volume (Lm3)	Comments/Notes
8" Insulated tailings lines	Tailings	8" Insulated discharge	varies	varies	2600	8	12	6 to 8 inch	1	0	2,600	0	0	247	
4" HDPE lines	Tailings				2800	4	8	6 to 8 inch	1	0	2,800	0	0	118	
4" Insulated HDPE lines					1400	4	4	0.75 to 4 inch	1	1,400	0	0	0	15	
4" Steel lines, heat traced					450	4	4	0.75 to 4 inch	1	450	0	0	0	4	
8" HDPE					1400	8	8	6 to 8 inch	1	0	1,400	0	0	45	
3: HDPE					600	3	3	0.75 to 4 inch	1	600	0	0	0	3	
10" Steel					240	10	10	10 to 18 inch	1	0	0	240	0	12	
16" HDPE					1500	16	16	10 to 18 inch	1	0	0	1,500	0	195	
24" HDPE					885	24	24	20 to 36 inch	0	0	0	0	885	0	
TOTALS:									8	**2,450**	**6,800**	**1,740**	**885**	**638**	

Notes:
Updated to match Table 6.1 in RCP

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B. Pipeline Infrastructure/Equipment

Area	Details	Quantity	Estimated Weight (kg)	Total Weight (kg)	Comments/Notes
Pumps		**15**			
Main Pit	Reclaim Pump	6	100	600	
Area 2	Dewatering Pumps	1	500	500	
Misc Pumps	Tsynami	7	200	1400	
W15 Sump	W15 Pump	1	200	200	
Barges/Misc. Equipment		**2**			
Main Pit	Reclaim Barge	1	2000	2000	
Main Pit	Power Shack	1	2000	2000	
TOTALS:				**6700**	

Notes:
1. Weights of pumping equipment are assumed
2. Equipment is assumed to be placed in semi-trailers for transport to Whitehorse for disposal/salvage.

C. Power Line and Substation Removal

				Off-site Disposal Trips				
Line	Power Line Length (km)	Number of power poles	Number of Transformers	Power Pole Trips	Power Line Reels	Transformer/ Powerline Trips	TOTAL TRIPS	Comments/Notes
Mill to Water Storage Pond	1.58	30	2	1.2	5.2	1.5		
Mill to Minto South Portal	1.58	30	2	1.2	5.2	1.5		
TOTALS:	**3.16**	**60**	**4**	**3**	**11**	**4**	**18**	

Notes:
1. Power line lengths and number of poles provide by Minto in file "powerpoles.dxf"

Worksheet 32- Underground

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

References (dwgs/plans): \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\!040_AutoCAD\1CM002-45_ClosureQuantities.dwg

A: Existing Portals

	Inputs							Calculated Quantities					
Description	Height (m)	Width (m)	Backfill Plug Thickness (m)	Backfill height above portal opening (m)	Backfill width outside of portal (m)	Backfill outside side slope (H:1V)	Relocation Cost Code for LHD	Underground Plug Volume (m^3)	Plug Volume in Front of Portal (m3)	Total Plug Volume (m^3)	Backfill LHD Productivity (m^3/hr)	Small Dozer Hours	Comments/Notes
Existing Portals													
Minto South Portal	5	7	15	5	12	3	R.001	788	1,800	2,588	159	16	
Minto North Portal	5	7	15	5	12	3	R.001	788	1,800	2,588	159	16	
Minto North Portal Egress	5	7	15	5	12	3	R.001	788	1,800	2,588	159	16	
Future Portals													
								0	0	0	#N/A	#N/A	

Notes:

B: Shafts and Vent Raises

	Inputs								Calculated Quantities												
Facility	Length (m)	Width (m)	Depth to competent material (m)	Concrete Cap Length (m)	Concrete Cap Width (m)	Concrete Slab Thickness (m)	Concrete Slab rebar layers	Backfill Thickness over cap (m)	Concrete Cap Perimeter (m)	Excavation Volume (m3)	Vent Raise Pipe Length (m)	I-beams required (W 250x33)	Total I-Beam Length (m)	Concrete Ring Wall Formwork (m2)	Concrete Ring Wall Rebar Length (m)	Ringwall Dowell Length (m)	Steel Decking (q-deck) (m2)	Total Rebar Length (m)	Total Concrete Volume (m3)	Backfill Volume (m3)	Comments/Notes
Existing shafts and vent raises																					
Area 118 Vent Raise	3	3	0.5	4	4	0.45	2	1	16	4.0	4	2	8	8	16	16	16	32	8.6	25	
Minto East	3	3	0.5	4	4	0.45	2	1	16	4.0	4	2	8	8	6	8	16	14	8.6	25	Assumed same dimentions as Area 118
Minto East 2	3	3	0.5	4	4	0.45	2	1	16	4.0	4	2	8	8	16	0	16	16	8.6	25	
Future shafts and vent raises																					
Minto South																					
Wildfire																					

CALCULATION INPUTS

Vent Raise design

I-Beam spacing (m):	2
Concrete ring wall height (m):	0.5
Concrete ring wall rebar spacing (into bedrock) (m):	1
Concrete ring wall rebar (into bedrock) length (m):	1
Ringwall dowel spacing at top of wall (m):	0.4
Ringwall dowel length (m):	0.4
Concrete slab rebar spacing:	0.2

Worksheet 33 - Waste Disposal

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

References (dwgs/plans): \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\I040_AutoCAD\1CM002-45_ClosureQuantities.dwg

A: Reagents

		Inputs					Calculated Quantities										
Area	**Reagents**	Solid/Liquid	Mass of Solid (kg)	Volume of Liquid (m^3)	Container Type	Disposal Fee (per container)	Number of drums	Number of tote bags	Number of Solid Bulk Carriers	Number of Small Tankers	Number of Large Tankers	Loading Time (hrs)	Trips Required - 55 Gal. Drums	Trips Required - Tote Bags	Disposal Fees ($)	**TOTAL TRIPS**	Comments/Notes
Mill Area	PAX	Solid	2,000		55 gal. drum	$0	7	0	0	0	0	1	0.1	0.0	$0		
	MIBC	Solid	3,652		55 gal. drum	$0	12	0	0	0	0	2	0.1	0.0	$0		
	MagnaFlocc 338	Solid	5,000		55 gal. drum	$0	17	0	0	0	0	3	0.2	0.0	$0		
	MagnaFlocc 155	Solid	500		55 gal. drum	$0	2	0	0	0	0	0	0.0	0.0	$0		
	Filter cleaning agent (HNC	Liquid		3.20	Liquid - 2,200 gal. tank	$0	0	0	0	1	0	4	0.0	0.0	$0		
Water Treatment Plant	Reagents	Solid	20,000		Tote bags	$0	0	15	0	0	0	8	0.0	0.4	$0		Assumes 15 totes remaining
MVF Reagent Tent	Reagents	Solid	20,000		Tote bags	$0	0	15	0	0	0	8	0.0	0.4	$0		Assumes 15 totes
TOTAL	**TOTAL**						38	30	**0**	**1**	**0**	**25**	**0.4**	**0.9**	**$0**	**2.2**	

Notes:
1. See details on container types below
1. Reagents assumed to be returned to supplier (no disposal fee)

B: Hazardous Materials

		Inputs					Calculated Quantities										
Area	**Description**	Solid/Liquid	Mass of Solid (kg)	Volume of Liquid (m^3)	Container Type	Disposal Fee (per container)	Number of drums	Number of tote bags	Number of Solid Bulk Carriers	Number of Small Tankers	Number of Large Tankers	Loading Time (hrs)	Trips Required - 55 Gal. Drums	Trips Required - Tote Bags	Disposal Fees ($)	**TOTAL TRIPS**	Comments/Notes
Mill Area	Misc. drums, oils, glycol &	Solid	20,000		55 gal. drum	$750	65	0	0	0	0	65	0.6	0.0	$48,750		
Camp Area	Misc. oils, cleaning supplie	Solid	100		55 gal. drum	$750	1	0	0	0	0	1	0.0	0.0	$750		
Fuel Farm Area	Misc. fuel containers/oil	Solid	500		55 gal. drum	$750	2	0	0	0	0	2	0.0	0.0	$1,500		
Explosives Plant	Misc. explosive material	Solid	2,000		Solid - Bulk	$0	0	0	1	0	0	4	0.0	0.0	$0		Assumed
Airport Laydown	Misc. fuel containers/oil	Solid	500		55 gal. drum	$750	2	0	0	0	0	2	0.0	0.0	$1,500		Assumed
Pelly Laydown	Misc. fuel containers/oil	Solid	500		55 gal. drum	$750	2	0	0	0	0	2	0.0	0.0	$1,500		Assumed
Minto South Portal	Misc. fuel containers/oil	Solid	500		55 gal. drum	$750	2	0	0	0	0	2	0.0	0.0	$1,500		Assumed
Airstrip Area	Recycling storage area ma	Solid	3,000		55 gal. drum	$750	10	0	0	0	0	10	0.1	0.0	$7,500		Assumed
TOTAL							84	0	**1**	**0**	**0**	**88**	**0.8**	**0.0**	**$63,000**	**1.8**	

Notes:
1. See details on container types below
2. Mill area parameters are assumed, and were adjusted to total $61,000 in the cost estimate to match a contractor quote from General waste management to MEL.

C: Metal Contaminated Soils

			Inputs				Calculated Quantities			
Area	**Description**	Disposal Location	Suspected Contaminated Area (m2)	Test Grid Spacing (m)	Assumed % Contaminated	Assumed Average Excavation Depth (m)	Delineation Test Pits Required	Confirmation Tests Required	Volume of Contaminated Soil (m3)	Comments/Notes
Crusher Area	Metal Contaminated Soils	UG	19,661	10	50%	0.3	197	197	2949	Suspected area is near processing/tailings/concentrate buildings
Mill Area	Metal Contaminated Soils	UG	19,822	10	50%	0.3	198	198	2973	Suspected area is near processing/tailings/concentrate buildings
TOTAL							395		5922	

Notes:
1. During investigation program, 2 soil tests completed per grid location (1 near surface and 1 at depth)

D: Hydrocarbon Contaminated Soils

			Inputs				Calculated Quantities			
Area	**Suspected location**	Disposal Location	Suspected Contaminated Area (m2)	Test Grid Spacing (m)	Assumed % Contaminated	Assumed Average Excavation Depth (m)	Delineation Test Pits Required	Confirmation Tests Required	Volume of Contaminated Soil (m3)	Comments/Notes
Fuel Storage Area	near fueling areas	Landfarm	1,644	10	15%	1.0	16	16	247	Entrance, fueling area, and near generators.
Mill area	near mechanic shop	Landfarm	550	10	15%	1.0	6	6	83	
Airstrip	allowance	Landfarm	500	10	15%	1.0	5	5	75	
Dundas area	allowance	Landfarm	500	10	15%	1.0	5	5	75	
Pelly laydown	allowance	Landfarm	1,644	10	15%	1.0	16	16	247	
TOTAL							48		725.7	

Notes:
1. During investigation program, 2 soil tests completed per grid location (1 near surface and 1 at depth)

E: Landfarm

		Inputs						Calculated Quantities						
Facility	**Location**	Total Volume to be treated (m3)	Maximum thickness (m)	Containment berm height (m)	Liner Protection Layer thickness (m)	Mixing events per year	Years mixing required	Landfarm Area (m2)	Landfarm Length/Width (m)	Landfarm Perimeter Length (m)	Containment Berm Volume (m3)	Liner Area (m2)	Total Mixing Events	Comments/Notes
Existing Landfarm	Airstrip	726				4.0	3.0	4265				4265	12	Overall areas provided by Minto
TOTAL								4265		0	0			

Notes:
1. During investigation program, 2 soil tests completed per grid location (1 near surface and 1 at depth)

F: Solid Waste Landfill

		Inputs						Calculated Quantities						
Facility	**Location**	Total Demolition Waste to be stored (LCM)	Assumed Percentage of fill to be added to provide compaction	Maximum fill thickness (m)	Landfill Side slopes (H:1V)	Landfill base length (m)	Landfill cover thickness (m)	Compacted Waste Volume (CCM)	Fill Volume (CCM)	Total Required Landfill Volume (CCM)	Calculated Landfill Volume (CCM)	Landfill Cover Volume	Revegetation Area (ha)	Comments/Notes
Solid Waste Landfill	Airstrip	12,920	30%	5	4	73	0.6	11,235	3,371	14,606	14712	5024	0.55	
TOTAL														

Notes:
1. During investigation program, 2 soil tests completed per grid location (1 near surface and 1 at depth)

Worksheet 34 - Water Conveyance/Storage Quantity Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

References (dwgs/plans): \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\!040_AutoCAD\1CM002-45_ClosureQuantities.dwg
Layer: For secondary conveyance channel - "Drainage Paths"

A: Primary Conveyance Channels

	Channel Parameters													X-Section Quantities					Total Quantities					
Area	Length (m)	Channel Base Width, Bc (m)	Side Slope, z (:1)	Average Final Channel Depth, Df (m)	Flow depth including Freeboard, D (m)	Excavation Volume User Over-ride (BCM)	Rip-Rap Thickness (m)	Bedding Layer Thickness (m)	Geotextile Layers	Liner? (Y/N)	Geosynthetic Wastage	Geosynthetic Anchor Trench Allowance (m)	Average Excavation Depth (m)	Average Excavated Area (m^2)	Rip-Rap Area (m^2)	Bedding Layer Area (m^2)	Geotextile Length (m)	Liner Length (m)	Excavation Volume (m^3)	Rip-Rap Volume (CCM)	Bedding Layer Volume (CCM)	Geotextile Area (m^2)	Liner Area (m^2)	Comments/Notes
WSP Breach Channel Restoration (pilot channel	250	2	2	2	1		0.5	0.3	1	No	10%	0.3	2.8	22.3	3.9	2.9	11.0	0.0	5,584	964	726	3,033	0	
Channel A1 (W15 to Main Pit)	83	10	2	10	1.5	40300	0.6	0	1	No	10%	0.3	n/a	485.5	10.9	0.0	20.3	0.0	40,300	906	0	1,851	0	1CM002-042/Jan. 16.
Channel A2	159	10	2	1.5	1.5	0	0.4	0	1	No	10%	0.3	n/a	0.0	7.1	0.0	19.3	0.0	0	1,126	0	3,373	0	1CM002-042/Jan. 16.
Channel A3	107	10	2	1.5	1.5	0	0.9	0	1	No	10%	0.3	n/a	0.0	17.0	0.0	21.8	0.0	0	1,823	0	2,561	0	1CM002-042/Jan. 16.
SUBTOTAL - Channel A																			40,300	3,855	0	7,785		
Channel B1 (W35 to Area 2 Pit)	72	7	2	2	1.35		0.4	0	1	No	10%	0.3	2.4	28.8	5.6	0.0	15.6	0.0	2,072	404	0	1,237	0	1CM002-042
Channel B2	52	7	2	2	1.23		0.6	0	1	No	10%	0.3	2.6	32.5	8.4	0.0	16.1	0.0	1,688	436	0	919	0	1CM002-042
Channel B3	83	7	2	10	1		0.9	0	1	No	10%	0.3	10.9	318.6	12.3	0.0	16.5	0.0	26,440	1,023	0	1,508	0	1CM002-042
Channel B4	56	7	2	10	1		0.9	0	1	No	10%	0.3	10.9	318.6	12.3	0.0	16.5	0.0	17,839	690	0	1,018	0	1CM002-042
SUBTOTAL - Channel B																			48,038	2,554	0	4,682	0	
Channel C (Main Pit to Main Access Road)	182	7	2	3	1.4		0.4	0	1	No	10%	0.3	3.4	47.6	5.7	0.0	15.8	0.0	8,656	1,037	0	3,171	0	1CM002-042
Channel D1 (Area 2 Pit to Main Pit Channel)	160	7	2	4	2.1		0.4	0	1	No	10%	0.3	4.4	70.4	7.0	0.0	19.0	0.0	11,256	1,112	0	3,339	0	1CM002-042
Channel D2	78	7	2	4	1.4		0.9	0	1	No	10%	0.3	4.9	84.4	13.9	0.0	18.3	0.0	6,583	1,087	0	1,571	0	1CM002-042
SUBTOTAL - Channel D																			17,840	2,199	0	4,910	0	
Channel E1 (Mill area)	278	7	2	2	2		0.4	0	1	No	10%	0.3	2.4	28.8	6.8	0.0	18.5	0.0	7,999	1,883	0	5,664	0	1CM002-042
Channel E2 to E6	647	7	2	2	2		0.9	0	1	No	10%	0.3	2.9	38.4	16.4	0.0	21.0	0.0	24,814	10,580	0	14,942	0	1CM002-042
Channel E7	33	7	2	2	2		1.2	0	1	No	10%	0.3	3.2	44.7	22.7	0.0	22.5	0.0	1,475	749	0	816	0	1CM002-042
Channel E8	46	7	2	2	2		0.9	0	1	No	10%	0.3	2.9	38.4	16.4	0.0	21.0	0.0	1,764	752	0	1,062	0	1CM002-042
Channel E9	150	7	2	2	2		2	0	1	No	10%	0.3	4	63.8	41.8	0.0	26.4	0.0	9,567	6,267	0	4,361	0	1CM002-042
Channel E10	20	7	2	2	2		0.9	0	1	No	10%	0.3	2.9	38.4	16.4	0.0	21.0	0.0	767	327	0	462	0	1CM002-042
SUBTOTAL - Channel E (SEE NOTE 3)																			46,386	20,558	0	27,307	0	
Tailings Diversion Ditch	420	0	6	0.5	0.8	0	1.6	0	1	No	10%	0.3	n/a	0.0	31.4	0.0	30.1	0.0	0	13,169	0	13,889	0	1CM002-042
Mill Valley Fill Wetland conveyance channels	160	4	2	1	1		0	0.3	1	Yes	10%	0.3	1.3	8.8	0.0	2.8	10.6	10.6	1,402	0	442	1,858	1,858	1CM002-042
Minto Creek Wetland By-Pass Channel	569	7	2	3	2		0.6	0	1	No	10%	0.3	3.6	52.1	10.5	0.0	19.5	0.0	29,668	5,950	0	12,212	0	
Minto North Spillway	250	7	2	3	1		0.6	0	1	No	10%	0.3	3.6	52.1	7.8	0.0	15.0	0.0	13,035	1,943	0	4,136	0	Allowance
Ridgetop North Outlet Channel (to Minto Crk)	290	3	2	2	1		0.9	0	1	No	10%	0.3	2.9	26.8	8.7	0.0	12.5	0.0	7,758	2,531	0	3,995	0	

Notes:
1. See channel drawing and parameter definitions below.
2. Channel A rip-rap thickness is a weighted average based on Jan. 16. 2018 design drawings.
3. 27,500 m3 of bulk fill required to form Channel E - this volume has been hardcoded into the cost estimate (A13.5). Volume was calculated in AutoCAD Civil3d

B: Secondary Conveyance Channels

	Channel Parameters										X-Section Quantities				Total Quantities				
Area	Length (m)	Channel Base Width, Bc (m)	Side Slope, z (:1)	Average Final Channel Depth, (m)	Rip-Rap Thickness (m)	Bedding Layer Thickness (m)	Geotextile Layers	Geotextile Wastage	Geosynthetic Anchor Trench Allowance (m)	Average Excavation Depth (m)	Average Excavated Area (m^2)	Rip-Rap Area (m^2)	Bedding Layer Area (m^2)	Geotextile Length (m)	Excavation Volume (m^3)	Rip-Rap Volume (CCM)	Bedding Layer Volume (CCM)	Geotextile Area (m^2)	Comments/Notes
Southwest Dump																			
SWD SC1 - Shallow grade	389	2	2	1	0.3	0	1	10%	0.3	1.3	6.2	2.2	0.0	8.6	2,398	842	0	3,661	See Note 1
SWD SC3 - Shallow grade	825	2	2	1	0.3	0	1	10%	0.3	1.3	6.2	2.2	0.0	8.6	5,085	1,785	0	7,764	
SWD SC4 - Shallow grade	276	2	2	1	0.3	0	1	10%	0.3	1.3	6.2	2.2	0.0	8.6	1,701	597	0	2,597	
SWD SC5 - Shallow grade	260	2	2	1	0.3	0	1	10%	0.3	1.3	6.2	2.2	0.0	8.6	1,603	563	0	2,447	
SWD SC1 - Steep grade	21	2	2	0.6	0.6	0.3	1	10%	0.3	1.5	8.1	3.7	2.5	9.7	171	78	53	225	
SWD SC2 - Steep grade	240	2	2	0.6	0.6	0.3	1	10%	0.3	1.5	8.1	3.7	2.5	9.7	1,953	888	604	2,570	
SWD SC3 - Steep grade	225	2	2	0.6	0.6	0.3	1	10%	0.3	1.5	8.1	3.7	2.5	9.7	1,831	832	566	2,409	
SWD SC4 - Steep grade	190	2	2	0.6	0.6	0.3	1	10%	0.3	1.5	8.1	3.7	2.5	9.7	1,546	703	478	2,034	
SWD SC5 - Steep grade	140	2	2	0.6	0.6	0.3	1	10%	0.3	1.5	8.1	3.7	2.5	9.7	1,139	518	352	1,499	
SUBTOTAL - South West Dump															**17,427**	**6,806**	**2,054**	**25,206**	
Main Waste Dump																			
MWD - Shallow grade	1051	2	2	1	0.3	0	1	10%	0.3	1.3	6.2	2.2	0.0	8.6	6,477	2,274	0	9,888	
MWD - Steep grade	490	2	2	0.6	0.6	0.3	1	10%	0.3	1.5	8.1	3.7	2.5	9.7	3,986	1,812	1,233	5,244	
SUBTOTAL - Main Waste Dump															**10,462**	**4,086**	**1,233**	**15,132**	
Reclamation Overburden Dump																			
ROD - Shallow grade	793	2	2	1	0.3	0	1	10%	0.3	1.3	6.2	2.2	0.0	8.6	4,886	1,715	0	7,460	
ROD - Steep grade	370	2	2	0.6	0.6	0.3	1	10%	0.3	1.5	8.1	3.7	2.5	9.7	3,007	1,367	930	3,956	
SUBTOTAL - Reclamation Overburden Dump															**7,893**	**3,083**	**930**	**11,416**	
Ridgetop Waste Dump																			
RWD - Shallow grade	93	2	2	1	0.3	0	1	10%	0.3	1.3	6.2	2.2	0.0	8.6	574	201	0	876	
RWD - Steep grade	43	2	2	0.6	0.6	0.3	1	10%	0.3	1.5	8.1	3.7	2.5	9.7	353	161	109	465	
SUBTOTAL - Ridgetop Waste Dump															**927**	**362**	**109**	**1,341**	
Dry Stack Tailings Storage Facility and Mill Valley Fill Buttress (Stage 1 and 2)																			
DSTSF - Shallow grade	692	2	2	1	0.3	0	1	10%	0.3	1.3	6.2	2.2	0.0	8.6	4,263	1,497	0	6,509	
MVFE - Shallow grade	505	2	2	1	0.3	0	1	10%	0.3	1.3	6.2	2.2	0.0	8.6	3,115	1,094	0	4,756	
DSTSF - Steep grade	322	2	2	0.6	0.6	0.3	1	10%	0.3	1.5	8.1	3.7	2.5	9.7	2,623	1,193	812	3,452	
MFVE - Steep grade	236	2	2	0.6	0.6	0.3	1	10%	0.3	1.5	8.1	3.7	2.5	9.7	1,917	872	593	2,522	
SUBTOTAL - DSTSF and MVFE															**11,919**	**4,655**	**1,405**	**17,239**	
TOTAL - Year 0 Estimate	7024																		
TOTAL - Peak Liability Estimate	7024																		

Notes:
1. SWD secondary channel lengths obtained from SRK project number 1CM002.031 (Progressive reclamation)
2. Channel lengths for all other dumps were based on SWD design scaled by area. 26.6 m of shallow sloped channel length per dump area (ha) were applied, and 12.4 m of steep sloped channel length area
3. 26.6 m of shallow sloped channel length per dump area (ha) was applied, and 12.4 m of steep sloped channel length per dump area (ha) was applied
4. Approx. dump areas used in the calculations are: SWD - 65.8 ha, MWD - 39.5 ha, ROD - 29.8 ha, DSTSF - 26 ha, and MVFE - 19 ha.

C1: Construct New Ponds/Sediment Basins

	Pond Parameters														Total Quantities					
Area	Pond Base Length (m)	Channel Base Width, Bc (m)	Side Slope, z (:1)	Final Pond Depth, Df (m)	Protected depth including Freeboard, D (m)	Excavation Volume User Over-ride (BCM)	Rip-Rap Thickness (m)	Rip-Rap Volume User Over-ride (BCM)	Bedding Layer Thickness (m)	Geotextile Layers	Liner? (Y/N)	Geosynthetic Wastage	Geosynthetic Anchor Trench Allowance (m)	Excavation Depth (m)	Excavation Volume (BCM)	Rip-Rap Volume (CCM)	Bedding Layer Volume (CCM)	Geotextile Area (m^2)	Liner Area (m^2)	Comments/Notes
Stilling Basin at MVFE Stage 2 Toe	15	5	2	2.5	2.5		0	75	0.3	1	No	10%	0.3	2.8	645	75.00	131	540	0	Rip-rap assumes an area of 75m2 at both the inlet and outlet with a thickness of 0.5m

Notes:
1. See channel drawing and parameter definitions below.

C2: Decommission Ponds/Sediment Basins

	Pond Parameters							Calculations			Total Quantities					
Name	Base Length (m)	Base Width (m)	Pond Side Slope (:1)	Pond Depth (m)	Sediment Depth (m)	Water Depth (m)	Liner Bulking Factor (See note 1)	Top Length (m)	Top Width (m)	Liner X-Section Width (mid-pond) (m)	Backfill Volume (m3)	Water Volume (m^3)	Sediment Volume (BCM)	Liner Area (m^2)	Liner Debris Volume (LCM)	Source, Comments/Notes
Mill Water Pond	43	2.5	4	5	0.5	4	10	83.0	42.5	43.7	7,088	4,366	101	3,630	54	Pond to be Filled In; dimension source: 2011 Annual Review plan drawing
Sewage Lagoon at IROD Laydown	35	25	3	3	0	0	n/a	53.0	43.0	44.0	4,488	0	0	2,331	#VALUE!	No liner present - dimensions from AutoCAD
W15 Sump							10	131.0	10.0	10.0				1,310	20	Liner Length from AutoCAD; width/depth assumed
Fuel Tank Farm	45	30	1	2	0.45	0	10	49.0	34.0	35.7	3,008	0	623	1,747	26	Farm dimensions from AutoCAD;
Waste Oil Tanker Containment	12	8	1	2	0.45	0	10	16.0	12.0	13.7	280	0	47	219	3	Farm dimensions from AutoCAD;

Notes:
1. Liner assumed to be 60mil HDPE (thickness 1.5mm); bulking factor estimated

D: Dams

	Dewatering Inputs				Dam Volumes								Calculations							
Name	Storage Capacity (m^3)	Pond Drawdown Rate (m^3/hr)	Seepage during drawdown (L/s)	Pump dewatering system install time (hrs)	Stripping Volume (BCM)	General backfill (CCM)	Upstream Shell (CCM)	Dam Core (CCM)	Filter Layers (CCM)	Downstream Shell (CCM)	Toe Berm (CCM)	Spillway Rip-rap (CCM)	Rip-rap % able to be reused	Time Required to dewater (days)	Total Dewatering Volume (m^3)	Rip-rap for Reuse (CCM)	Rip-rap to discard (CCM)	Total Dam Fill Vol	General Fill to Breach (CCM)	Dam Core to Breach (CCM)	Downstream Shell to breach (CCM)	Source, Comments/Notes
Water Storage Pond Dam	360,000	1080	30	4	33,031	10,740	30,449	29,883	31,630	43,005	6,013	5,800	50%	15	400,000	19,285	19,285	157,520	62,301	23,617	38,571	Max drawdown rate: 300L/s (1,080m3/hr)

Notes:
1. WSP max drawdown rate from "Dam Stability Analysis - Water Retention Dam Reservoir Dewatering (EBA 2008)
2. Dam volumes from asbuilt report (EBA 2008)
3. Upstream shell is residuum, downstream shell is rip-rap sized material, the core is fine-grained clayey material, the filters are screened/crushed materials

E: Intake Structures

	Pond Parameters							Total Quantities		
Name	Quantity	Length (m)	Width (m)	Height (m)	Depth to Competent Foundation (m)	Over-excavation Allowance (m) (See note 1)	Excavation Side Slopes (H:1)	Excavation Volume (BCM)	Backfill volume (CCM)	Source, Comments/Notes
Main Pit Pre-cast concrete intake structure	1	15	4	3	3	2	1	2,450	2,450	Based on pre-cast concrete panel wall manufactured by Precon
Area 2 Pit pre-cast concrete intake structure	1	15	4	3	5	2	1	4,453	4,453	Based on pre-cast concrete panel wall manufactured by Precon

Notes:
1. The perimeter around the intake structure assumed to be over excavated around the perimeter to for worker access during installation.

F: Passive Treatment System/Wetlands

	Valley Fill Parameters				Wetland Cell Parameters												Calculated Quantities							
Area	Length (m)	Average Width (m)	Average Fill Thickness (m)	Valley Fill Volume User Over-ride (CCM)	Number of cells	Base of organics Length (m)	Base of organics width, Bc (m)	Organic Thickness (m)	Final Pond Depth, Df (m)	Side Slope, z (:1)	Excavation Volume User Over-ride (BCM)	Organic Volume User Over-ride (BCM)	Geotextile Layers	Liner? (Y/N)	Geosynthetic Wastage	Geosynthetic Anchor Trench Allowance (m)	Valley Fill Volume	Cell Excavation Depth (m)	Cell Excavation Volume (BCM)	Organic Media Volume (CCM)	Subgrade preparation area (m2)	Geotextile Area (m^2)	Liner Area (m^2)	Comments/Notes
Stilling Basin at MVFE Stage 2 Toe	400	40	4.5		7	41	15	1	1.3	2			1	Yes	10%	0.3	72,000	2.3	14,459	5,122	9,273	10,201	10,201	

Notes:
1. See channel drawing and parameter definitions below.
2. The domains/formulas in the channel drawing below are adjusted in the Wetlands calculations as follows: Wetland media is the equivalent of 'water' (blue), the protection layer is the equivalent to bedding/filter layer (green), and rip-rap was not included in the calculations.

Worksheet 35 - Annual Water Treatment Cost Calculations

Project: Minto Mine Closure Cost Estimate - RCP 2018-01
Project No.: 1CM002.045
Client: Minto Explorations Ltd.
Date of Submission: February 6, 2018
File Location: \\VAN-SVR0\Projects\01_SITES\Minto\1CM002.045_ClosureCosts\RCP2018-1_CostEstimate\

A: Active Treatment

Adjustment Factors

Inflation factor 1.03 Applied to 2015 site costs to 2017 Dollars based on Bank of Canada Inflation Calculator

Operating Costs

Item	Crew/ Unit	Hours	Materials ($)	Labor Cost ($/hr)	Equipment Cost ($/hr)	Total Materials ($)	Total Labor ($/hr)	Total Equip ($/hr)	Total Cost ($)	Comments/Notes
Contractor Equipment	1	1			$38,831	$0	$0	$38,831	$38,831	
Reagents	1	1	$47,277			$47,277	$0	$0	$47,277	
Filters	1	1	$63,963			$63,963	$0	$0	$63,963	
Operating Parts	1	1	$198,584			$198,584	$0	$0	$198,584	
Power	1	1	$146,260			$146,260	$0	$0	$146,260	
O&M Supplies	1	1	$16,377			$16,377	$0	$0	$16,377	
Membranes	1	1	$39,346			$39,346	$0	$0	$39,346	
Labour	1	1241		$52.53		$0	$65,194	$0	$65,194	Based on plant availability 34%, 10hr/day.
TOTAL - Annual Active Treatment Operating Costs						**$511,807**	**$65,194**	**$38,831**	**$615,832**	

NOTES:
1. Based on site costs - 2015, which have been found to be a typical year for the plant operation.
2. Source: "2015 - WTP Costs.xlsx' supplied by Minto in April 2016.

Capital Replacement Costs

Item	Crew/ Unit	Hours	Materials ($)	Labor Cost ($/hr)	Equipment Cost ($/hr)	Total Materials ($)	Total Labor ($/hr)	Total Equip ($/hr)	Total Cost ($)	Comments/Notes
Mechanical/Electrical Parts	1	1	$61,903			$61,903	$0	$0	$61,903	
Pipes & Fittings	1	1	$25,029			$25,029	$0	$0	$25,029	
Replacement Labour	1	80		$52.53		$0	$4,203	$0	$4,203	Hours assumed.
TOTAL - Annual Active Treatment Operating Costs						**$86,932**	**$4,203**	**$0**	**$91,135**	

NOTES:
1. Based on site costs - 2015
2. Source: "2015 - WTP Costs.xlsx' supplied by Minto in April 2016.

B: Passive Treatment

Wetland maintenance

Item	Crew/ Unit	Hours	Materials ($)	Labor Cost ($/hr)	Equipment Cost ($/hr)	Total Materials ($)	Total Labor ($/hr)	Total Equip ($/hr)	Total Cost ($)	Comments/Notes
Excavator	1	80		$44	$113	$0	$3,551	$9,073	$12,624	
Std. Dump Truck	2	80		$44	$25	$0	$7,014	$3,956	$10,970	
Misc. supplies	1	1	$10,000			$10,000	$0	$0	$10,000	Reveg, bacteria, etc.
Labour	1	40		$36		$0	$1,421	$0	$1,421	
TOTAL - Annual Active Treatment Operating Costs						**$10,000**	**$11,986**	**$13,029**	**$35,015**	

NOTES:
1. Assumed costs for earthwork repairs/cell repair and hydraulic adjustments.