REPORT ON THE 1998 WORK PROGRAM

CAM CLAIMS 1 - 146

LIVINGSTONE AREA

WHITEHORSE MINING DISTRICT, YUKON

NTS 105 E/8

by

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INTRODUCTION:

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1998 was the second year of work on the hardrock CAM Claims in the Livingstone placer camp. During 1997, soil sampling along the north rims of the placer creeks produced anomalous gold, copper, and arsenic values in areas where shear zones were expected. It was decided to confine work on the north rims of the creeks because they had little disturbance from placer workings, they would have less overburden than the south sides of the creeks because the glaciers traveled over the area from the south or southeast, and the north side of the creeks face south so would thaw more quickly and have fewer permafrost problems. The shear zones seen on the ground as well as in aerial photographs was confirmed with the use of ground VLF-EM surveys in several areas.

A detailed soil sampling grid with coincident gold, copper, and arsenic values which also contained shear zones confirmed with ground VLF-EM in the area of an old adit on the north side of Livingstone Creek was excavated with a total of 5 bulldozer trenches in early May, 1998. An additional 6 trenches (Ron, Windlass 1 & 2, and Mandy, Mandy West, and Mandy Southwest) were excavated at the same time (See trench drawings). These trenches are in the area of the headwaters of Summit Creek and on the ridge between Summit and Lake Creeks (See CAM Claims, 1998 Work Program – in pocket).

The general prospecting, rock and soil sampling performed during 1998 were chiefly directed at extending the strike lengths of known mineralization further toward the north from the creek rims explored during 1997. Good soil and rock

samples obtained during 1998 resulted in two days of additional bulldozer trenching in late October. One trench was placed over a soil sample having a high gold value (326 ppb.) north of Cottoneva Creek and two trenches were located north of Lake Creek in areas where rock and soil samples had returned high gold values (See trench drawings).

Knowledge of the property was aided by visits from Kennecott, Viceroy, YTG, and DIAND geologists. This report has been prepared to describe the 1998 work program and provide conclusions and recommendations for further work on the CAM Claims.

LOCATION, ACCESS AND CLAIMS:

The original 142 CAM Claims were staked in 1997 to cover 5 of the 6 placer creeks which make up the Livingstone placer camp. Mining of these creeks is still occurring 100 years after it first started. The CAM Claims are located on NTS Map Sheet 105 E/8 and are centered at approximately Latitude 61[°] 19' N; Longitude 134[°] 17' W within the Whitehorse Mining District, Yukon (See Part of Claim Map 105 E/8 Quartz). An additional 4 claims were staked just east of the main block in May, 1998. These claims were staked to protect a trench, called the Ron Trench, excavated that month. The trench exposed a segment of a sheared quartz vein containing copper and gold values. The claims are owned 50% each by Larry W. Carlyle and Max Fuerstner of Whitehorse, Yukon.



The Livingstone Creek area is accessed by a 75-mile winter road from Lake Laberge. Several air strips exist in the Livingstone area so access is usually via fixed-wing aircraft from Whitehorse; approximately 50 air miles to the south southwest. The main Livingstone air strip is 4000 feet long and has had DC-3 and Caribou aircraft landed on it. The extensive placer mining which has taken place in the area has resulted in cat trails existing up most of the creeks within the claim block. These trails enable easy access to most areas by all-terrain vehicles.

The claims cover areas which extend from the rim extending along the eastern side of the Big Salmon Fault at an elevation of approximately 900 metres (2,950 ft.) to above timberline near the top of the hills above the headwaters of the creeks at an elevation of approximately 1500 metres (4,920 ft.). The claims are on rounded to steeply sloping hills; the creek canyons have the steepest slopes. Vegetation consists of black spruce, pine, willow and buckbrush.

CLAIM NAME GRANT NUMBERS EXPIRY DATE

CAM 1 - 126	YB 97530 - YB 97655	May 16, 1999
CAM 127 - 142	YC 07943 - YC 07958	July 22, 1999
CAM 143 - 146	YC 08748 - YC 08751	May 19, 1999

REGIONAL GEOLOGY:

The geology and the placer gold deposits of the Livingstone Creek area were first described by McConnell in 1901. Regional geological mapping was carried out by Cockfield, Lees, and Bostock between 1929 and 1934. This work resulted in Map 372 A being issued in 1936 (See Part of Map 372 A). Most of the camp was mapped as Unit 1, Precambrian quartzite, schists, limestone, gneiss, and greenstone. Along the headwaters of most of the creeks, they mapped a sheared granodiorite as Unit 2. This unit is unique and not found elsewhere on the map sheet. Further east they mapped a large zone of peridotite, hornblendite, and serpentine as Unit 10. A small stock of Unit 11, probably a Cretaceous granite, granodiorite, monzonite, or diorite was mapped at the headwaters of Little Violet Creek (See Part of Map 372 A).

The regional geology was reinterpreted by Tempelman-Kluit in 1977-1979 (See Part of G.S.C. O.F. 1101). This interpretation identified the Big Salmon Fault, down which the South Big Salmon River flows and into which the placer creeks drain. During this mapping, Tempelman-Kluit identified the Teslin Fault (4 - 6 miles west of the Livingstone camp) as the ancient western margin of North America. Tempelman-Kluit obtained more accurate age dating for the rocks of the area; and has mapped most of the rocks as Carboniferous and/or Permian dark green, fine-grained amphibolite and amphibolitic greenstone (CP_{Av}). He has mapped Unit CP_{Ag}, a dioritic to quartz dioritic augen amphibole gneiss, in almost exactly the same location as the Unit 2 from the 1936 map.



The rocks west of the Teslin Fault (also known as the Teslin Suture) were pressed against and over the original North America during the Early Cretaceous. His theory postulated that this action would cause the rocks east of the Big Salmon Fault to be raised in reverse faulted thrust blocks.

Tempelman-Kluit's westerly dipping subduction zone with North American rocks in the footwall and accreted arc terrane and oceanic rocks in its hanging wall has been reinterpreted. This reinterpretation, which has been developing from the mid-1980's to 1997, considers the Teslin zone as a zone of ductile thrusting, which includes thrust sheets of North American affinity and accreted rocks that have been complexly folded and displaced northeastward and then folded again. Rather than marking the western limit of rocks of North American origin, the zone is most likely underlain by North American basement that extends westward beneath the Intermontane Belt. Two facts strongly support this model over that of Tempelman-Kluit:

- the same metamorphosed stratigraphies can be traced along a strike length of at least 20 km. This would not be possible in the more chaotic jumble of rock blocks expected from collapsing hangingwall rocks into a subduction zone.
- most of the rocks in the area have green schist or amphibolite grade metamorphism. Rocks in a subduction zone would most probably have eclogite or blue schist grade metamorphism.

In the new model; Devonian-Mississippian granites and Permian intrusives are deformed, while Late Triassic to Early Jurassic plutons are undeformed; this would put the age of deformation and metamorphism between Late Permian and Late Triassic. Proponents of this model, suggest renaming the Teslin Suture



Zone, the Teslin Tectonic Zone. Rocks within the Teslin Tectonic Zone are correlated with sedimentary and volcanic rocks of the Yukon Tanana terrane and oceanic crustal rocks of the Slide Mountain terrane. Yukon Tanana terrane rocks range in age from Devonian to Permian. After their deformation and cooling, the Slide Mountain terrane rocks were emplaced over them along low-angle, post-metamorphic faults. In the Big Salmon Range (just north of Livingstone), the Teslin Tectonic Zone is 20 km. wide. Both Slide Mountain and Yukon Tanana rocks contain steeply dipping fabrics, unlike their counterparts in the rest of the Yukon and Alaska.

The steep north-south striking D'Abbadie fault has generally been taken to represent the eastern margin of the Teslin Zone. It is most probably a narrow zone of brittle deformation reflecting a period of upper crustal normal faulting superimposed on the ductile deformation which had occurred earlier. Last Peak granite has been dated at 98 Ma. and, on the basis of contact and structural relationships, is interpreted to have intruded while the D'Abbadie fault zone was active. Dextral shearing and gentle NW plunging of the stratigraphy are also believed to have occurred at this time.

PROPERTY GEOLOGY:

Rock outcrop is limited on the CAM Claims. Exposure is generally restricted to creek canyons and to west-facing rock bluffs running parallel to the Big Salmon Fault. Outcrop is also frequently found in strong depressions (notches) seen cutting across ridges and extending for considerable distances along the hillsides above timberline. Most of the rocks seen on the property are metasediments of green schist or amphibolite metamorphic grade. These metasediments are dominated by strongly contorted biotite-chlorite-quartz schist. Some of the schist contains intercalated thin- and medium-bedded quartzite. Small very discontinuous patches of white to grey limestone are located in the area. Caliche is found frequently in fractures and along the bedding or schistosity of the rocks. Aerial magnetometer surveys show a couple of small magnetic highs. The first extends across Livingstone Creek in the area of Sheen's Gulch; the second extends across the lower end of Summit Creek. A ground magnetic survey over the area in Livingstone Creek showed the source to be a chlorite schist containing small magnetite crystals. This rock is believed to be the metamorphic equivalent of a basic volcanic.

White bull quartz-calcite veins or boudins are found in the depressions mentioned earlier. The quartz-calcite veins have widths from 4 inches up to 4 feet, but are most commonly 1 to 2 feet wide. The gold mineralization is believed to be in or associated with these quartz-calcite veins or boudins. The depressions are faults or shear zones having a strike between $320^{\circ} - 340^{\circ}$ Az. and appear to have westerly dips between $65 - 75^{\circ}$. The faults have widths of 2 - 10 metres but are usually 3 - 5 metres wide. The approach of a fault or shear zone is recognized by the alteration of biotite schist to chlorite schist then to sericite schist. This is accompanied by stronger shearing and increased light to dark brown iron oxide as a fault is approached. The faults are more closely spaced directly east of the Big Salmon Fault and more widely spaced further toward the east.

MINERALIZATION:

After having read Stroink and Friedrich (1992) the gold mineralization was believed to be in or associated with the quartz-calcite veins or boudins. Very little mineralization other than trace oxidized pyrite was seen in any of the quartz exposed on surface.

Pyrite, galena, and copper and silver sulphides were not present in the quartz from the ridges and gullies. This mineralization was only seen in vein quartz from the Horseshoe Adit (See 1998 Work Program, Cam Claims - in pocket) and from a quartz vein in the placer workings on Livingstone Creek.

At the beginning of the 1998 season, mineralization on the property was expected to be concentrated within fault or shear zones visible on the aerial photographs and within the "notches" seen on the ridges and along the hillsides as mentioned earlier. Trenching and soil sampling done in the shear zones during 1998 have shown that many of the shears contain thick deposits of glacial till or return low gold values. It is now thought that the shears provided the "plumbing system" for the mineralizing fluids which flowed out into fractures and shears in the surrounding country rock -- the main shears then rehealed. It is still probable that economic grade mineralization will be concentrated within "pockets" within the minor fractures and shears as well as the major shears.

1998 WORK PROGRAM:

The work undertaken in 1998 consisted of bulldozer trenching, rock sampling and soil sampling. The work was chiefly directed toward extending known mineralized zones north along strike from their locations on the north sides of the creeks.

Trenching

Bulldozer trenching was undertaken at two times during 1998; in early May and again in late October. The prime focus for the trenching in May was to expose the bedrock source of mineralization north of the old Horseshoe Adit located with soil sampling and VLF-EM during 1997. Five trenches were excavated in the area to investigate this mineralization as well as to find out if mineralization was located within shear zones. This trenching located several narrow (approx. 1 ft. wide) zones of galena mineralization containing gold values up to 0.938 opt. (32.16 g/t) approximately 50 metres northwest of the adit in Trench 3W (See 1998 Work Program, Cam Claims - in pocket; as well as appropriate trench drawings). Two zones within this trench possess economic gold values over mineable widths:

• a zone at 71 metres from the east end averages 1.44 g/t over 1.47 metres

• a zone at 89 metres from the east end averages 4.89 g/t over 2.84 metres A single shear zone was located in Trench 4, Trench 3E, and Trench 2; the highest gold value obtained from the shear zone was 43 ppb. Au (Sample T4 S1).

While demobilizing the bulldozer, 6 additional trenches were excavated. The first of these was the Ron Trench located within a "notch" east of the ridge between











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TRENCH 4 SCALE; 1 cm = \sim 5 m







SCALE; $1 \text{ cm} = \sim 5 \text{ m}$



Sheen's Gulch and Summit Creek (See 1998 Work Program, Cam Claims - in pocket; as well as appropriate trench drawing). A 1 metre wide crushed sericitequartz vein associated with a thin-bedded graphitic quartzite-graphite schist zone was exposed along approximately ten metres of its strike. The vein has trace pyrite and malachite mineralization containing gold values up to 0.1 g/t. Two Windlass Trenches were excavated in a "notch" where old-timers had sunk a shaft on which their windlass still existed. Deep overburden was encountered and rock samples returned low gold values (See 1998 Work Program, Cam Claims - in pocket; as well as appropriate trench drawing). Three Mandy Trenches were excavated in the next "notch" toward the west (See 1998 Work Program, Cam Claims - in pocket; as well as appropriate trench drawing). Again, deep overburden was encountered, and no shears were located. Sample Mandy-1, a piece of float, returned a gold value of 98 ppb.

In late October, another three trenches were excavated. The first was located approximately 200 metres north of Cottoneva Creek on soil sample C98-3 which returned a gold value of 326 ppb. Au (See 1998 Work Program, Cam Claims - in pocket; as well as appropriate trench drawing). This was the second highest gold value obtained from a soil sample on the property. The highest was 898 ppb. Au (0.9 g/t) obtained in 1997 from Sample SC'-4 on Summit Creek. Although a shear zone was exposed directly below the soil sample location, rock samples taken from the shear returned disappointing gold grades. During trenching, it was discovered that the soil sample was located on an esker of glacial till, so probably represents a transported anomaly.





The last two trenches were excavated north of Lake Creek. They were called Trench 1 and Road Cut Trench (See 1998 Work Program, Cam Claims - in pocket: as well as appropriate trench drawing). Trench 1 was cut along a 1997 soil sample line on which Samples LAK-4, 5, and 6 returned gold values of 46, 149 and 87 ppb. respectively. Although anomalous gold values were located in the shear zone exposed in the trench, the best values (169.9 ppb. Au [0.17 g/t] over 20.5 metres) were obtained from the trench's east end (footwall ?) silicified sericite-quartz schist. The Road Cut Trench was excavated approximately 12 metres directly north of a road cut from which the writer had obtained a grab sample of highly silicified sericitic schist with quartz stringers which ran 262 ppb. Au but from which Vicerov geologists had obtained a grab sample which ran 3110 ppb. Au (3.1 g/t.). Samples from the entire trench averaged 60.1 ppb. Au (0.06 a/t.) over 29.0 m. However, a 5.0 m. segment returned 190 ppb. Au (0.19 g/t). The structures located within the two trenches appear to be along strike; the structure would, therefore, have a strike exposure of 90 - 100 metres.

The great difficulty in obtaining repeat gold values, ie: Ron Trench and Lake Creek road cut samples strongly suggests that nugget effect is present in the bedrock mineralization.

See Appendix A for Trench Rock Sample and Miscellaneous Rock Sample Values and Descriptions.

Rock Sampling

Rock samples, excluding trench samples, were not very numerous. The greatest number of samples were taken on Livingstone Creek primarily from the placer mine cut in the canyon on lower Livingstone Creek in an area known as Blake's Bar (See 1998 Work Program, Cam Claims - in pocket). During placer mining in this area, a 4-6 foot wide andesitic(?) dyke with large (up to 3 cm.) plagioclase (?) phenocrysts was broken through. The dyke was steeply dipping and has a strike of 300[°] to 340[°] Az. It is assumed that the old-timers thought the dyke was rim rock. Upon breaking through it, a zone of up to 30 feet wide of virgin pay gravels were discovered behind it. When the rim rock was reached; a wide shear zone of altered sericite schist-quartz having strong iron oxide alteration was located. This shear and a few miscellaneous samples from the area were sampled (See Rock Sample Table).

Sample SR - 1 was taken approximately ¼ mile up the Summit Falls Road from the Livingstone townsite. Samples L+IV - 1, LV-1, 2, and 3 were taken in the area of lower Little Violet Creek. Sample LV-1 is a garnet-actinolite skarn and is believed to be the first skarn mineralization located on the claims. Samples M-1, and 2 were taken from the canyon area of Mendocina Creek. The location of these samples has been placed on the 1998 Work Program, Cam Claims Map in pocket. A table of these rock sample descriptions and mineral values has been included as Appendix A.

Soil Sampling

During 1998, soil sampling was again an important tool for the exploration of the CAM Claims. On May 4, 1998, 14 soil samples in two lines were taken north of the line of samples taken in 1997 at Cottoneva Creek (See 1998 Work Program, Cam Claims Map - in pocket). These samples were taken at 20 metre intervals along two lines separated by 50 metres. Both lines crossed a prominent "notch" directly east of Max Fuerstner's present camp and were obtained to further investigate the gold potential within these shear zones. The poor gold values obtained from the samples (Numbered Cot-1 to Cot-14 [See Soil Sample Table]) may have been the result of their having been taken too early in the season before the ground was sufficiently thawed.

On June 7, 1998, two more groups of soil samples were taken across two "notches" located on the ridge between Sheen's Gulch and Summit Creek. The first group were taken in a "notch" directly along strike from the lower segment of Sheen's Gulch where it runs into Livingstone Creek (See 1998 Work Program, Cam Claims Map - in pocket). These samples were prefixed by "SG". The second group were taken in a "notch" further toward the east; and were prefixed with "SGE". Although both "notches" were thought to not contain much overburden, the more easterly one had almost 50 foot vertical rock walls for much of its length. The presence of a thin layer of overburden is considered important since thick overburden may prevent a geochemical response to soil sampling (See Soil Sample Table). The low gold values obtained from these samples may again be the result of the ground being too frozen. On June 8, 1998, 20 soil samples were taken in two lines with a sample spacing of 20 metres starting 50 metres north of the 1997 soil line at the Lake Creek "notch". There is 50 metres between the 1998 lines (See 1998 Work Program, Cam Claims Map - in pocket). These samples were prefixed with "Lk98" (See Soil Sample Table). These samples may also have been taken before the ground had sufficiently thawed.

On September 15, 1998, 13 soil samples were taken at 25 metre spacings at Little Violet Creek. These samples started at a point approximately 200 metres north of the bottom of the mine cut and followed a line of 80^o Az. This direction resulted in the final sample (V98-13) being approximately 800 feet north of the edge of the mine cut (See 1998 Work Program, Cam Claims Map - in pocket). It was decided that this was too far from the edge of the cut so an additional line was started from the same point as the first line but samples were taken along a line at 90^o Az [Samples V98-14 to 25] (See Soil Sample Table). While doing this work, the 1997 line of soils was located running along the northern edge of the mine cut where the overburden is a minimum of 70 feet thick. It is for this reason that those samples are thought not to have returned significant gold values.

On September 17, 1998, 16 soil samples were taken approximately 50 metres north of the 1997 line at Summit Creek. Samples were spaced at 25 metre intervals and were prefixed "S98" (See 1998 Work Program, Cam Claims Map in pocket and Soil Sample Table). Samples from the 1998 line returned gold

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Summit Creek

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Sample #	Location	Au (ppb)	Ag (ppm)	Cu (ppm)	As (ppm)	Pb (ppm)	Zn (ppm)
				-			
S98 - 1	0 + 00 E	< 5	< 0.1	18	48	13	36
S98 - 2	0 + 25 E	< 5	< 0.1	28	40	4	30
S98 - 3	0 + 50 E	< 5	< 0.1	10	34	4	30
S98 - 4	0 + 75 E	< 5	< 0.1	16	35	6	31
S98 - 5	1 + 00 E	< 5	< 0.1	21	34	9	39
S98 - 6	1 + 25 E	5	0.2	52	59	14	63
S98 - 7	1 + 50 E	< 5	< 0.1	23	46	11	42
S98 - 8	1 + 75 E	< 5	< 0.1	29	47	10	49
S98 - 9	2 + 00 E	< 5	< 0.1	33	64	14	56
S98 - 10	2 + 25 E	12	< 0.1	39	48	13	52
S98 - 11	2 + 50 E	13	0.1	40	57	10	57
S98 - 12	2 + 75 E	14	0.2	51	67	15	84
S98 - 13	3 + 00 E	8	0.2	47	59	14	68
S98 - 14	3 + 25 E	7	< 0.1	48	52	13	58
S98 - 15	3 + 50 E	5	0.2	91	66	9	72
S98 - 16	3 + 75 E	8	0.3	108	74	4	73

Sheen's Gulch

Sample #	Location	Au (ppb)	Ag (ppm)	Cu (ppm)	As (ppm)	Pb (ppm)	Zn (ppm)
SG - 1	L1 0+80 E	< 5	0.4	12	15	9	42
SG - 2	L1 0+60 E	5	0.2	15	18	14	50
SG - 3	L1 0+40 E	< 5	0.2	9	6	14	25
SG - 4	L1 0+20 E	< 5	0.4	7	8	< 2	17
SG - 5	L1 0+00 E	< 5	0.3	8	· 8	5	32
SG - 6	L2 0+00 E	< 5	0.3	27	20	27	101
SG - 7	L2 0+20 E	< 5	0.2	8	6	2	10
SG - 8	L2 0+40 E	7	0.3	19	12	7	34
SG - 9	L2 0+60 E	6	0.2	18	15	5	35

Sheen's Gulch East

Sample #	Location	Au (ppb)	Ag (ppm)	Cu (ppm)	As (ppm)	Pb (ppm)	Zn (ppm)
SGE - 1	L1 0+00 E	< 5	0.4	10	8	< 2	27
SGE - 2	L1 0+18 E	8	0.3	15	9	< 2	27
SGE - 3	L2 0+11 E	< 5	0.2	3	< 5	< 2	13
SGE - 4	L2 0+00 E	6	< 0.1	6	< 5	< 2	12
SGE - 5	L3 0+00 E	< 5	< 0.1	1	< 5	< 2	6
SGE - 6	L3 0+12 E	5	0.4	13	19	11	91
SGE - 7	L4 0+15 E	< 5	< 0.1	1	< 5	< 2	6
SGE - 8	L4 0+00 E	6	0.3	18	18	6	42

Lake Creek

.

Sample #	Location	Au (ppb)	Ag (ppm)	Cu (ppm)	As (ppm)	Pb (ppm)	Zn (ppm)
Lk98 - 1	L1 0+00 E	< 5	0.2	4	< 5	< 2	10
Lk98 - 2	L1 0+20 E	< 5	0.1	8	5	< 2	8
Lk98 - 3	L1 0+40 E	< 5	0.2	8	< 5	< 2	12
Lk98 - 4	L1 0+60 E	< 5	0.2	10	10	9	29
Lk98 - 5	L1 0+80 E	< 5	0.2	7	10	6	23
Lk98 - 6	L1 1+00 E	< 5	0.2	5	9	5	19
Lk98 - 7	L1 1+20 E	< 5	< 0.1	2	< 5	< 2	7
Lk98 - 8	L1 1+40 E	< 5	< 0.1	4	< 5	< 2	12
Lk98 - 9	L1 1+60 E	< 5	0.2	24	7	2	14
Lk98 - 10	L1 1+80 E	31	0.3	19	5	6	18
Lk98 - 11	L2 1+80 E	6	0.3	18	12	11	27
Lk98 - 12	L2 1+60 E	< 5	0.2	22	13	8	36
Lk98 - 13	L2 1+40 E	< 5	0.3	22	< 5	< 2	33
Lk98 - 14	L2 1+20 E	5	0.2	9	< 5	4	19
Lk98 - 15	L2 1+00 E	< 5	0.2	19	13	7	36
Lk98 - 16	L2 0+80 E	9	0.3	4	< 5	< 2	10
Lk98 - 17	L2 0+60 E	< 5	0.3	8	16	14	25
Lk98 - 18	L2 0+40 E	< 5	0.3	3	< 5	< 2	12
Lk98 - 19	L2 0+20 E	9	0.3	13	10	5	34
Lk98 - 20	L2 0+00 E	< 5	0.2	4	< 5	7	13

Cottoneva Creek

Sample #	Location	Au (ppb)	Ag (ppm)	Cu (ppm)	As (ppm)	Pb (ppm)	Zn (ppm)
Cot - 1	L1 0+00 E	5	< 0.1	1	< 5	< 2	9
Cot - 2	L1 0+20 E	5	< 0.1	4	< 5	< 2	13
Cot - 3	L1 0+40 E	7	< 0.1	9	8	4	23
Cot - 4	L1 0+60 E	< 5	0.1	6	6	< 2	15
Cot - 5	L1 0+80 E	< 5	< 0.1	7	< 5	6	19
Cot - 6	L1 1+00 E	7	< 0.1	8	6	< 2	22
Cot - 7	L2 1+00 E	10	< 0.1	35	27	7	43
Cot - 8	L2 0+80 E	6	< 0.1	10	12	5	23
Cot - 9	L2 0+60 E	< 5	< 0.1	9	8	4	22
Cot - 10	L2 0+40 E	< 5	< 0.1	7	< 5	< 2	18
Cot - 11	L2 0+20 E	8	< 0.1	11	11	6	26
Cot - 12	L2 0+00 E	7	< 0.1	7	6	3	26
Cot - 13	L2 0+20 W	5	< 0.1	1	< 5	< 2	5
Cot - 14	L2 0+40 W	5	< 0.1	4	< 5	< 2	7
Sample #	Location	Au (ppb)	Ag (ppm)	Cu (ppm)	As (ppm)	Pb (ppm)	Zn (ppm)
C98 - 1	L3 0+00 E	< 5	< 0.1	21	29	5	28
C98 - 2	L3 0+25 E	< 5	< 0.1	19	31	8	31
C98 - 3	L3 0+50 E	326	< 0.1	21	56	8	45
C98 - 4	L3 0+75 E	16	0.1	17	34	7	32
C98 - 5	L3 1+00 E	9	< 0.1	15	37	11	38
C98 - 6	L3 1+25 E	< 5	< 0.1	10	30	8	35
C98 - 7	L3 1+50 E	14	0.1	16	51	5	41
<u> </u>	L3 1+75 E	< 5	0.1	15	31	8	33
C98 - 9	L3 2+00 E	< 5	0.1	32	48	8	48
C98 - 10	L3 2+25 E	< 5	< 0.1	12	30	4	30
C98 - 11	L3 2+50 E	< 5	0.1	9	27	5	28
C98 - 12	L3_2+75 E	< 5	< 0.1	22	41	< 2	31
C98 - 13	L3_3+00 E	7	0.3	77	98	18	125
C98 - 14	L3 3+25 E	< 5	< 0.1	11	26	3	90
C98 - 15	L3 3+50 E	5	< 0.1	35	58	8	58
C98 - 16	L3 3+75 E	7	0.2	81	62	8	62
C98 - 17	L3 4+00 E	< 5	< 0.1	12	34	6	34
C98 - 18	L3 4+25 E	7	< 0.1	46	62	13	62

Miscellaneous Cottoneva Creek

Sample #	Location	Au (ppb)	Ag (ppm)	Cu (ppm)	As (ppm)	Pb (ppm)	Zn (ppm)
			,				
CT - 1	~200 m. E	6	0.3	28	23	9	37
	of Camp						_
	_						
CG - 1	South of	9	0.3	14	16	10	30
	Creek						
CG - 2		25	0.2	27	22	6	46
CG - 3	**	9	0.4	29	13	< 2	43

Little Violet Creek

Sample #	Location	Au (ppb)	Ag (ppm)	Cu (ppm)	As (ppm)	Pb (ppm)	Zn (ppm)
V98 - 1	0+00 E	< 5	< 0.1	18	22	4	33
V98 - 2	0+25 E	< 5	< 0.1	14	26	3	21
V98 - 3	0+50 E	< 5	0.1	13	43	8	23
V98 - 4	0+75 E	< 5	0.1	36	53	7	34
V98 - 5	1+00 E	< 5	0.1	14	33	4	26
V98 - 6	1+25 E	< 5	< 0.1	10	23	4	23
V98 - 7	1+50 E	< 5	< 0.1	8	26	6	21
V98 - 8	1+75 E	< 5	< 0.1	25	32	7	33
V98 - 9	2+00 E	< 5	< 0.1	12	27	7	23
V98 - 10	2+25 E	< 5	< 0.1	37	53	10	45
V98 - 11 ·	2+50 E	< 5	< 0.1	24	38	7	33
V98 - 12	2+75 E	< 5	< 0.1	11	26	9	27
V98 - 13	3+00 E	11	0.1	21	46	12	54
V98 - 14	0+25 E	< 5	0.1	16	26	6	27
V98 - 15	0+50 E	45	0.1	23	30	5	30
V98 - 16	0+75 E	9	< 0.1	38	33	5	38
V98 - 17	1+00 E	8	< 0.1	13	18	6	23
V98 - 18	1+25 E	< 5	< 0.1	11	22	10	25
V98 - 19	1+50 E	9	< 0.1	22	36	7	34
V98 - 20	1+75 E	< 5	< 0.1	8	20	5	19
V98 - 21	2+00 E	8	0.1	30	46	8	40
V98 - 22	2+25 E	< 5	< 0.1	17	27	8	29
V98 - 23	2+50 E	< 5	< 0.1	19	29	5	29
V98 - 24	2+75 E	16	< 0.1	26	33	9	32
V98 - 25	3+00 E	7	0.1	67	54	14	69
values up to only 14 ppb.; however, these values are along strike from the 898 ppb. Au value obtained in 1997.

The final line of soil samples for the season were taken approximately 200 metres north of Cottoneva Creek. They were taken at 25 metre spacings along an azimuth of 80⁰ and were given the prefix "C98" (See 1998 Work Program, Cam Claims Map - in pocket and Soil Sample Table). Sample C98 - 3 of these samples returned a 326 ppb. Au value and was subsequently trenched (See **1998** *Context of the season were taken at 25 metre spacings along an appropriate trench drawing*).

CONCLUSIONS:

- 1. Gold mineralization is present within the country rock of the historic placer creeks in the Livingstone Creek area.
- 2. For the moment, economic gold grades over mineable widths have only been located over a strike length of approximately 200 metres in the area of the old Horseshoe Adit on the north side of Livingstone Creek. However, sub-economic gold grades have been located in bedrock on the next three creeks toward the north: Summit, Lake, and Cottoneva Creeks.
- 3. Silicification is strongest at Livingstone Creek and appears to get progressively weaker toward the north. Skarn mineralization was located for the first time at Little Violet Creek. By the time Mendocina Creek is reached, at the north end of the property, relatively fresh sheared limestone is located (See 1998 Work Program, Cam Claims - in pocket).
- 4. The significant bismuth values obtained from rock samples T3W S1, T3W - S3, and T3W - S5 and the lack of it within soil samples taken over the area; as well as the low gold values obtained from the shear zone exposed directly below the 326 ppb. Au value soil sample at Cottoneva Creek suggests that soil sample anomalies may have experienced glacial transport.

CARI VI

5. The relatively recent discovery of large tonnage intrusive hosted, low grade gold deposits in the Yukon (Brewery Creek, Dublin Gulch) and Alaska (Fort Knox, Pogo) suggest a closer look should be given to the sheared granodiorite (Unit 2) of Map 372 A and the dioritic to quartz dioritic augen amphibole gneiss (Unit CP_{Ag}) of G.S.C. O.F. 1101. These units were mapped in the same locations, within the claim block, at the headwaters of the placer creeks by Cockfield, Lees, and Bostock between 1929 and 1934; as well as by Tempelman-Kluit in 1977-1979.

RECOMMENDATIONS:

- 1. Extending the known zone of mineralization at the Horseshoe Adit both north (uphill) and south (downhill) along strike with VLF-EM and soil sampling would advance the property significantly.
- 2. Bulk samples of approximately 100 to 150 pounds (45.5 68.2 kg.) should be obtained from the mineralized zone at the Horseshoe Adit and perhaps another area (possibly mineralized rock from Lake Creek) for petrographic, mineralogical, and metallurgical testing.
- 3. An attempt should be made to geologically map the property to:
 - investigate its potential for large tonnage intrusive hosted, low grade gold mineralization
 - determine if some feature, as yet unknown, controls whether the gold mineralization is of economic or sub-economic grade
 - investigate the extent of the newly discovered skarn rocks and their mineral potential
- 4. The direction and distance of potential glacial transport, discovered during 1998, should be determined.



STATEMENT OF COSTS:

TOTAL	\$ 34,897.54
Report Writing	<u>\$ 1,500,00</u>
Office Supplies (Photocopying, paper, etc.)	\$ 193.08
Field Supplies (Flagging, bags, hip chain twine, etc.)	\$ 200.00
Miscellaneous Fuels & Oil	\$ 200.00 TO PRAV
ATV Rental (2 weeks @ \$125./wk)	\$ 250.00
Room & Board (53 person/days @ \$35/day)	\$ 1,855.00
Air Charters	\$ 1,818.40 1 W. CARLYLE
Assaying	\$ 3,828.56
Geologist Field Work (24 days @ \$300/day)	\$ 7,200.00 Jan 1 Las
Bulldozer Trenching (96.5 hrs @ \$185./hr)	\$ 17,852.50 3 0 ¹⁰
	GEOLOGIA

NOTE: These costs do not include any costs incurred on the property from the property visits made by Kennecott, Viceroy, YTG, and DIAND geologists.

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STATEMENT OF QUALIFICATIONS

- I, LARRY W. CARLYLE, do certify:
- 1. That I am a professional geologist; resident at 74 Tamarack Drive, Whitehorse, Yukon Y1A 4Y6.
- 2. That I hold a B. Sc. Degree in geology from the University of British Columbia (1970).
- 3. That I am a Fellow of the Geological Association of Canada (F 4355).
- 4. That I am a Registered Professional Geologist in the Association of Professional Engineers, Geologists, and Geophysicists of the Province of Alberta (41097).
- 5. That I have practiced my profession as a mine and exploration geologist for over twenty years.
- 6. The conclusions and recommendations in the attached report are based on work I performed or supervised on the property, and on a review of the references cited.

DATED at Whitehorse, Yukon, this // day of December, 1998.



APPENDIX A

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TRENCH ROCK SAMPLE AND MISCELLANEOUS ROCK SAMPLE VALUES AND DESCRIPTIONS

.

Adit Trench 1

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb).	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
T1 S2	12' E Adit	0.5	234	1.9	244	19	274	25	< 2	White qtz lenses. Very strong Wad. Tr Pbs, Chalc, & Py. Strong limonite. Tr goethite.
T1 S3	16 m W of Adit	0.5	< 5	0.2	4	< 5	< 2	8	< 2	Blocky, white vein quartz. Limonite & Wad in fract. No visible sulphides.
T1 S4	6 m W of Adit	3.5	11	0.6	41	10	42	54	< 2	Friable ser. schs + qtz ztringers. Strong iron oxide. No visible sulphides.
T1 S5	2.5 m W of Adit	1.0	8	< 0.1	20	9	73	32	⁻ < 2	Fract. vuggy, white vein qtz Strong limonite & goethite. No visible sulphides.
T1 S6	1.5 m W of Adit	1.7	14	0.4	31	13	83	32	< 2	Friable ser. schs with weak iron oxide. Some graphitic zones No visible sulphides
T1 S7	0.3 m E of Adit	1.2	256	2	63	413	169	23	< 2	Crushed & fract. vein qtz + friable ser.schs Strong limonite, minor wad & goethite. No visible sulphides.
T1 S8	1.5 m E of Adit	4.9	19	0.5	29	< 5	70	53	< 2	Friable, iron stained ser. schs. Strong vein qtz lenses. No visible sulphides.

Sample T1 S4 - T1 S8 == 0.038 g./ 12.3 m.

Adit Trench 2

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
T2 S1	23 m off road	2.0	18	< 0.1	32	< 5	10	68	< 2	Shear Zone - Highly sheared & gougy grey ser-qtzite schs. Minor limonite.
T2 S2	25 m off road	2.0	28	< 0.1	31	11	15	57	< 2	Shear Zone - Highly sheared & gougy grey ser-qtzite schs. Minor limonite.

Adit Trench 3E

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
T3E S1	W edge shear	2.0	7	0.2	27	5	11	56	< 2	Shear Zone Limonitic sericite schs gouge.
T3E S2	Next to East	2.0	12	< 0.1	33	9	8	62	< 2	Dark grey sericitic (?) schs gouge.
T3E S2	Quartz float	Grab	< 5	0.1	9	< 5	< 2	8	< 2	Large quartz boulder which fell into trench Tr vuggy,white vein qtz ? Strong lim-wad in fractures. No visible sulphides.

Adit Trench 3W

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description			
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)				
T3W S1	71.3 m from E end	0.21	9.77*	94.7	73	< 5	6427	4	119	<u>* in grams.</u> White quartz lense. Limonite & goethite in fract. <1% PbS.			
T3W S2	70.0 m from E end	0.26	254	0.4	23	< 5	29	21	< 2	Gougy dark brn. limonite stained sericite-quartz schist.			
T3W S3	89.0 m from E end	0.26	32.2*	100	86	< 5	15110	4	173	<u>* in grams. Vugg</u> y quartz lenses. Tr py Limonite & goethite f. f.			
T3W S4	90.4 m from E End	0.18	22	0.7	13	< 5	82	8	< 2	Apparently barren white qtz. Limonite & wad f.f.			
T3W S5	89.3 m from E End	1.1	4918	35.5	126	< 5	3430	17	34	Friable, iron stained sericite schist. Weak white qtz lenses. No visible sulphides.			
T3W S6	88.7 m from E End	1.3	85	0.6	19	< 5	85	21	< 2	Sheared & gougy limonite stained sericite schist. Weak qtz lenses. No visible sulphides.			
Sampl	Sample T3W S4 - T3W S6 == 4.89 g./2.8 m Sample T3W S1 - T3W S2 == 1.44 g./1.5 m												

Adit Trench 4

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
T4 S1	15 m E of road	2.0	43	< 0.1	45	20	17	48	< 2	Lt. brn. limonitic sericitic gouge
T4 S2	17 m E of road	2.0	29	< 0.1	35	15	14	56	< 2	Lt. grey-brn. limonitic sericitic gouge
T4 S3	19 m E of road	2.0	15	< 0.1	29	< 5	12	51	< 2	Lt. brn. limonitic sericitic gouge

Ron Trench

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
Ron - 1	Vein	1.0	33	1.9	432	121	75	130	< 2	Crushed sericitic-argillic quartz schist Strong iron & manganese staining. Tr py, malachite & ZnS (?).
Ron - 2	Vein	1.0	5	0.2	141	23	11	23	< 2	As above
P134015	Vein	1.0	45	1.0	195	42	44	12	< 2	As above. <u>Viceroy</u>
80887	Vein	1.0	105	5.2	382	130	164	130	< 2	As above. <u>Kennecott</u>

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Windlass Trench 1

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Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
Windlass S1	29 - 32 m from E End	3.0	21	< 0.1	47	21	16	83	< 2	Orange-red iron oxide gouge.
Windlass S2	45 m from E End	Grab	9	< 0.1	21	22	< 2	6	< 2	White bull vein quartz.

Windlass Trench 2

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
Windlass S3	10.4 m to 19.0 m	8.6	20	< 0.1	43	< 5	5	52	< 2	Sericite schist, some zones of graphite schist + weakly banded quartzite

Mandy Trench 2

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	-
Mandy - 1	Float	Grab	98	0.5	16	< 5	36	6	< 2	Compact, tr vuggy, silicified sericitic schist. < 1% iron & wad f.f. Tr oxidized py crystals. Tr black, sub-metallic mineral (biotite ?).

Cottoneva Trench

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
CT - 1	17.8 m from E End	Grab	< 5	< 0.1	31	15	3	18	< 2	1.5 ft. wide white-grey blocky bull qtz vein. Weak iron oxide f.f. No visible sulphides.
CT - 2	17.3 m from E End	2.0	< 5	0.2	59	46	10	88	< 2	Friable graphite (biotite ?) schist. 1/16" iron oxide specks (py ?). Red-brn iron oxide f.f. Some gouge & calcite.
СТ -3	15.3 m from E End	2.0	8	0.2	58	44	11	85	< 2	Limy, friable graphitic schist. Strong It. brn. to red-brn iron oxide. Weakly gougy.
CT - 4	13.3 m from E End	2.0	5	< 0.1	51	34	16	69	< 2	Limy, friable graphitic schist. Strong lt. brn. to red-brn iron oxide. Weakly gougy.
CT - 5	11.3 m from E End	2.0	5	0.2	35	67	17	56	< 2	Limy, friable graphitic schist. Strong It. brn. to red-brn iron oxide. Weakly gougy. Some minor quartz.
CT - 6	9.3 m from E End	2.0	8	0.2	25	121	9	42	< 2	Limy, friable graphitic schist. Strong It. brn. to red-brn iron oxide. Weakly gougy. Some minor quartz.
CT - 7	7.3 m from E End	1.3	7	< 0.1	43	223	11	70	< 2	Limy, friable graphitic schist. Strong It. brn. to red-brn iron oxide. Weakly gougy. Some minor quartz.

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Cottoneva Trench

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
CT - 8	5.3 m from E End	2.0	12	0.6	73	44	96	85	< 2	Blocky graphitic (biotite ?) schist. Weakly limy. Weaker iron oxide f.f. Minor gouge.
CT - 9	3.3 m from E End	2	12	0.4	63	54	31	76	< 2	Blocky graphitic (biotite ?) schist. Weakly limy. Weaker iron oxide f.f. Minor gouge.

Lake Creek Trench 1

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
LT - 1	53 m from E End	5.0	26	0.3	21	9	7	15	< 2	Blocky qtzite-graph schs. Strong iron & wad in f.f.
LT - 2	48 m from E End	3.3	43	0.2	30	5	9	21	< 2	V. M.? Crushed & gougy ser-qtz schs. Strong iron oxide. Weak wad. No visible sulphides.
LT - 3	44.7 m from E End	5.1	62	0.6	15	6	42	19	< 2	Blocky qtzite-graph schs. Mod. lt. brn. iron oxide. Tr wad & chlorite.

Lake Creek Trench 1 (Continued)

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
LT - 4	39.6 m from E End	2.0	61	0.2	32	18	14	21	< 2	Vein Shear. Crushed & gougy ser-qtz schs. Strong red-brn iron oxide. Minor wad & Chlorite.
LT - 5	37.6 m from E End	2.0	44	< 0.1	33	19	10	23	< 2	Vein Shear. Crushed & gougy ser-qtz schs. Strong red-brn iron oxide. Minor wad & Chlorite.
LT - 6	35.6 m from E End	2.0	39	0.2	38	34	10	67	< 2	Vein Shear. Crushed & gougy ser-qtz schs. Strong red-brn iron oxide. Minor wad & Chlorite.
LT - 7	33.6 m from E End	2.0	13	< 0.1	70	9	7	20	< 2	Limy, blocky qtzite-graphite schist. Gougy V.M. along south side of trench.
LT - 8	31.6 m from E End	3.1	18	< 0.1	29	8	2	15	< 2	Limy, blocky qtzite-graphite schist.
LT - 9	28.5 m from E End	5.0	31	0.2	61	12	6	35	< 2	Strongly friable limy sericitic-qtz schs. Strong orange-brn iron oxide.
LT - 10	22.5 m from E End	5.0	107	0.4	50	7	20	38	< 2	Strongly friable limy sericitic-qtz schs. Strong orange-brn iron oxide.

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Lake Creek Trench 1 (Continued)

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
LT - 11	17.5 m from E End	5.0	235	0.5	61	< 5	8	37	< 2	Strongly friable limy sericitic-qtz schs. Strong orange-brn iron oxide.
LT - 12	12.5 m from E End	5.0	74	0.4	48	7	7	30	< 2	Strongly friable limy sericitic-qtz schs. Strong orange-brn iron oxide.
LT - 13	7.5 m from E End	5.5	255	0.5	45	14	13	32	< 2	Strongly friable limy sericitic-qtz schs. Strong orange-brn iron oxide.
Samples LT - 10 to LT - 13 == 0.17g./ 20.5 m										

Road Cut Trench

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
LT - 14	E End	5.0	58	0.3	38	< 5	6	34	< 2	Highly silicified sericite schist with quartz stringers & lenses up to 1.5 ft. wide. Weakly limy. Strong red & orange-brn iron oxide. Tr fresh py. Minor wad in f.f. ZnS ?
LT - 15	5.0 m to W	5.0	22	0.2	19	8	3	14	< 2	Highly silicified sericite schist with quartz stringers & lenses up to 1.5 ft. wide. Weakly limy. Strong red & orange-brn iron oxide. Tr fresh py. Minor wad in f.f. ZnS ?
LT - 16	10.0 m to W	5.0	190	0.6	46	7	21	19	< 2	Highly silicified sericite schist with quartz stringers & lenses up to 1.5 ft. wide. Weakly limy. Strong red & orange-brn iron oxide. Tr fresh py. Minor wad in f.f. ZnS ?
LT - 17	15.0 m to W	5.0	11	0.2	15	5	5	12	< 2	Highly silicified sericite schist with quartz stringers & lenses up to 1.5 ft. wide. Weakly limy. Strong red & orange-brn iron oxide. Tr fresh py. Minor wad in f.f. ZnS ?
LT - 18	20.0 m to W	5.0	27	0.3	33	9	9	17	< 2	Highly silicified sericite schist with quartz stringers & lenses up to 1.5 ft. wide. Weakly limy. Strong red & orange-brn iron oxide. Tr fresh py. Minor wad in f.f. ZnS ?

Road Cut Trench (Continued)

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Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
LT - 19	25.0 m to W	4.0	51	0.2	29	7	12	29	< 2	Highly silicified sericite schist with quartz stringers & lenses up to 1.5 ft. wide. Weakly limy. Strong red & orange-brn iron oxide. Tr fresh py. Minor wad in f.f. ZnS ?
Whole Trench Averages 0.06 g./ 29.0 m										

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Livingstone Creek

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
L - 1	Blake's Bar	Grab	< 5	0.1	42	54	11	62	< 2	Grey to black fine to medium grained andesite ? Large white plag. phenos up to 1/2" long. Partially altered (saussuritized ?) Small (up to 1/8") blk-grn pyroxene (?) phen No visible sulphides.
L - 2	Blake's Bar	Grab	< 34	0.2	36	49	6	58	< 2	Green-brn chloritic & iron rich gouge. Seems to be mixed in with dyke and qtz veins. Seems to rest against dyke H.W.
L - 3	Blake's Bar	Grab	34	0.1	22	< 5	< 2	9	< 2	Fractured white qtz vein with strong lt. brn- orange f.f. of iron oxide. > 1% py cubes Possible Po, Chalco, & PbS.
L - 4	"	Grab	40	0.8	34	108	21	38	< 2	<u>Grab by Ken Galambos.</u> He called it a skarn. I think it is a f.g. grey silicified limestone. Tr fresh py. Weak brn iron oxide f.f.
L - 5	North Cut Rim	2.4	10	0.2	37	< 5	10	23	< 2	Crushed sericitic schs with qtz stringers. Strong ornage-brn iron oxides. Tr py & wad.
L - 6	11	1.4	19	0.4	111	160	39	54	< 2	Crushed & gougy graphitic-ser. schs. (Chlorite ?) Minor iron oxide. No visible mineralization.

Livingstone Creek (Continued)

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
L - 7	North Cut Rim	1.7	8	0.2	41	31	20	42	< 2	Highly crushed & contorted ser. schs. Strong orange-brn iron oxide. Tr qtz & gouge.
L - 8	"	1.8	8	0.2	14	< 5	12	42	< 2	Highly crushed ser. schs. Strong qtz lenses with < 1% fresh py. Strong orange- brn iron oxide.
L - 9	II	1.6	8	0.2	14	15	4	44	< 2	Strongly fract. ser schs. Less qtz than L-8. Strong orange-brn iron oxide. Mod. wad f.f.
L - 10	U	1.7	8	0.2	28	9	11	44	< 2	Crushed, weakly gougy ser schs. Strong orange-brn iron oxide. Weak qtz stringers. Tr wad.
L - 11	"	3.0	8	< 0.1	48	28	11	68	< 2	Highly fract. & crushed ser schs. Minor qtz lenses & wad.
L - 12	300 ft. to E	Grab	7	0.3	24	30	10	43	< 2	Calcite cemented gravels just above bedrock.
L - 13	South side of creek	Grab	9	0.7	25	19	7	67	< 2	Graphite schs with minor yellow-orange iron oxide f.f.

Summit Creek

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
SR - 1	LivSum Trail	Grab	< 5	0.1	3	< 5	3	13	< 2	~ 1/4 mile up trail from Livingstone Ck. Grey qtzite with 1/4 - 2 1/2" ribbons of white qtz through it. No visible sulphides.

Little Violet Creek

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
L + I.V1	South Cut edge	2.0	5	0.4	160	< 5	10	59	< 2	Fract., sheared & gougy sericitic qtzite. Strong red-orange iron oxides. Tr oxidized py ? No other visible sulphides.
LV - 1	~ 1/4 mi. N of Ck.	Grab	< 5	< 0.1	2	21	2	28	< 2	Sugary textured brn garnet & actinolite skarn. Tr epidote. No visible mineralization.
LV - 2	North Cut edge	Grab	< 5	0.2	28	25	4	18	< 2	Banded skarn (?). Epidote serpentine (?) Oxidized py crystals up to 1/8".
LV - 3	11	Grab	7	< 0.1	49	38	10	77	< 2	Skarn (?) Epidote-garnet-ser. schs. Some iron oxide. Strongly fract. @ high angle. W dip. No visible mineralization. iron oxide f.f.

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Mendocina Creek

Sample #	Location	Width	Au	Ag	Cu	As	Pb	Zn	Bi	Description
		(m)	(ppb)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	
M - 1	Top of canyon	Grab	< 5	0.1	1	< 5	11	10	< 2	<u>From near biotite schist contact.</u> Sheared sugary textured white-lt. grey limestone. Weakly silicified & banded. No visible sulphides.
M - 2	South of creek opposite canyon	Grab	< 5	0.1	14	50	5	59	< 2	Biotite-garnet-qtz schist with 2-3" qtz stringers. Tr pyrite.

APPENDIX B

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ANALYTICAL CERTIFICATES

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Certificate of Analysis

Page 1

	Larry Carlyle
Livingstone	Assays

Livingstone F	tssay.s	Certified by
Sample #	Au ppb	()
- CT 1		
CT-2	<0	
r CT-3	8	
r CT-4	5	
r CT-5	5	
CT-6	ρ	
r CT-7	7	
CT-8	12	
r CT-9	12	
r L-4	40	
de de g - 1 - 5	10	
1	19	
	8	
L-8	8	
r L-9	8	
- 1 ₋₁₀	Q	
	8	
 -12	7	
r L-13	9	
r LT-1	26	
	40	
	43	
r	61	
	44	
r LT-6	39	
la de dedi redevide de la companya de la comp	10	
	10 10	
	31	
LT-10	107	
LT-11	235	



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Page 2

Larry Carlyle

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	Sample #	Au ppb	U U
	LT-12 LT-13 LT-14 LT-15 LT-16	74 255 58 22 190	
	LT-17 LT-18 LT-19 LV-2 LV-3	11 27 51 <5 7	
F	SR-1 M-2	<5 <5	
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V	10,	1998		Page	1 of	2
		Fax	(604)	879-7898		
		Phon	e (604)	879-7878		
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Client Projec	: Nort t: W.O.	ther 56	n Ana 25	lytical	Labo	ratori	es	4	12 Sa 42=1	mpl e Pulp	es							[1	19513:45:	58:891	11098	נ	Out: In :	Nov 10 Nov 05	F 199 199	ax 98 98	(604)	879-78 Page Sectio	198 0n 1
Sampl	e Name		Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm	Hg ppm	Mo T ppnn pp	1 Bi m ppm	Cd ppm	Co ppm	Ni ppm	Ba W ppm ppm	Cr ppm	V ppm	Mn ppm	La Sr ppm ppm	Zr ppm	Sc ppm	Ti X	A1 لا	Ca ¥	Fe X	Mg X	X K	Na X	P X
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CT - CT - CT - CT - L -	6 7 8 9 4		0.2 < 0.6 0.4 0.8	25 43 73 63 34	9 11 96 31 21	42 70 85 76 38	121 223 44 54 108	~ ~ ~ ~ ~	~ ~ ~ ~ ~	3 2 2 2 2 2	<	5.2 6.4 6.2 6.8 7.0	7 12 13 14 8	32 52 67 47 23	60 < 72 < 112 < 113 < 22 <	70 46 51 48 59	20 28 31 41 6	877 968 659 432 674	8 153 9 151 26 43 20 25 3 32	2 2 4 3 5	4 6 5 1	< < < 0.01 <	0.44 0.51 0.84 1.38 0.20	3.47 4.10 1.81 0.61 1.58	2.42 3.18 2.95 3.33 3.40	0.92 0.92 0.58 0.77 0.68	0.09 0.11 0.17 0.14 0.12	< < < < <	0.04 0.04 0.07 0.06 0.01
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L - L - L - L - LT -	10 11 12 13 1		0.2 < 0.3 0.7 0.3	28 48 24 25 21	11 11 10 7 7	44 68 43 67 15	9 28 30 19 9	~ ~ ~ ~ ~	~ ~ ~ ~ ~	6 5 2 6 2	~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~	6.2 6.6 5.0 3.5 2.1	22 18 12 4 4	44 38 52 26 10	47 < 752 < 107 < 61 < 45 <	38 55 71 34 55	10 42 36 5 3	879 431 456 180 604	12 66 16 38 8 47 3 25 20 127	12 8 6 10 12	5 4 3 2 1	<pre> </pre>	0.37 0.99 0.96 0.18 0.26	2.50 0.89 6.33 2.24 4.33	3.00 3.26 2.22 L.68 L.05	1.07 0.69 1.22 1.01 0.11	0.18 0.21 0.11 0.12 0.09	<pre> </pre>	0.06 0.08 0.05 0.04 0.01
LT - LT - LT - LT - LT -	2 3 4 5 6		0.2 0.6 0.2 < 0.2	30 15 32 33 38	9 42 14 10 10	21 19 21 23 67	5 6 18 19 34	~ ~ ~ ~ ~	~ ~ ~ ~ ~	3 2 2 3 3	~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~	2.9 2.0 2.9 2.5 7.5	6 4 5 26	14 9 16 11 46	47 < 48 < 50 < 47 < 136 6	48 64 56 56 91	5 6 11 7 53	551 552 743 507 868	22 130 20 75 27 49 30 32 47 18	10 11 11 10 13	1 2 3 2 11	< < < 0.03	0.34 0.33 0.60 0.56 1.53	3.70 2.93 2.53 1.47 0.64	L.43 L.05 L.38 L.32 B.72	0.16 0.29 0.26 0.18 0.79	0.12 0.10 0.10 0.10 0.29	0.01 0.01 0.01 0.01 <	0.02 0.01 0.02 0.02 0.03
LT - LT - LT - LT - LT -	7 8 9 10 11		< < 0.2 0.4 0.5	70 29 61 50 61	7 2 6 20 8	20 15 35 38 37	9 8 12 7 <	~ ~ ~ ~ ~	~ ~ ~ ~ ~	1 2 2 1 3	<	3.1 2.1 4.6 4.8 5.8	6 5 9 13 14	11 10 18 25 26	62 < 36 < 51 < 45 <	50 59 40 38 44	4 6 9 20	356 430 334 599 586	31 38 20 57 26 26 37 22 27 26	10 9 9 11 12	1 1 1 2 4	~ ~ ~ ~ ~	0.42 0.28 0.64 0.49 0.41	1.11 2.16 1.81 1.32 2.45	L.63 L.01 2.32 2.61 2.88	0.12 0.12 0.27 0.24 0.25	0.12 0.11 0.13 0.17 0.14	0.01 0.01 < <	0.02 0.01 0.02 0.03 0.03
LT - LT - LT - LT - LT -	12 13 14 15 16		0.4 0.5 0.3 0.2 0.6	48 45 38 19 46	7 13 6 3 21	30 32 34 14 19	7 14 < 8 7	~ ~ ~ ~ ~	< < < < <	4 2 1 3 2	<	4.2 4.5 3.7 2.2 3.3	7 10 5 4 4	16 19 10 8 9	55 < 64 < 74 < 46 < 40 <	43 37 50 74 46	6 5 5 4	425 456 451 523 686	22 18 34 25 15 31 11 24 14 107	12 11 8 11 16	2 2 1 1 2	< < < < <	0.39 0.57 0.33 0.20 0.19	2.92 1.98 2.32 2.05 4.32	2.16 2.24 1.80 1.01	0.16 0.13 0.59 0.49 0.49	0.13 0.16 0.08 0.06 0.06	<pre> </pre>	0.02 0.03 0.02 0.01 0.01
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Min Limit

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CERTIFICATE OF ANALYSIS iPL-98K1195

2036 Columbia Cimet Vancouver, B.C Canada V5Y 3E1

Phone (604) 879-7878 Fax (604) 879-7898

INTERNATIONAL PLASMA LABORATORY LTD.		
: Northern Analytical Laboratories	42 Samples	

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Sample Name	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm	Hg ppm	Mo ppm	Tl Bi ppm ppm	Cd ppm	Со ppm	Ni ppm	Ba W ppm ppm	Cr ppm	V ppm	Mn ppm	La ppm	Sr ppm	Zr ppm	Sc ppm	Ti X	A1 *	Ca X	Fe X	Mg X	K	Na X	P X	
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Min Limit Max Reported* Method



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Certificate of Analysis

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Page 1

Larry Carlyle

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	Sample #	ppb	
S	C98-1	<5	
S	C98-2	<5	
s	C98-3	326	
S	C98-4	16	
s	C98-5	9	
S	C98-6	<5	
S	C98-7	14	
S	C98-8	<5	
s	C98-9	<5	
S	C98-10	<5	
S	C98-11	<5	
S	C98-12	<5	
S	C98-13	7	
S	C98-14	<5	
S	C98-15	5	
S	C98-16	7	
S	C98-17	<5	
S	C98-18	7	
S	S98-1	<5	
S	S98-2	<5	
s	S98-3	<5	
S	S98-4	<5	
S	S98-5	<5	
S	S98-6	5	
S	S98-7	<5	
s	S98-8	<5	
S	S98-9	<5	
S	S98-10	12	
S.	S98-11	13	
S	S98-12	14	
L			



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Page 2

Larry Carlyle

	Au	
Sample #	ppb	_
s S98-13	8	
s S98-14	7	
s S98-15	5	
s S98-16	8	
s V98-1	<5	
s V98-2	<5	
s V98-3	<5	
s V98-4	<5	
s V98-5	<5	
s V98-6	<5	
s V98-7	<5	
s V98-8	<5	
s V98-9	<5	
s V98-10	<5	
s V98-11	<5	
s V98-12	<5	
s V98-13	11	
s V98-14	<5	
s V98-15	45	
s V98-16	9	
s V98-17	8	
s V98-18	<5	
s V98-19	9	
s V98-20	<5	
s V98-21	8	
s V98-22	<5	
s V98-23	<5	
s V98-24	16	
s V98-25	7	
L-1	<5	
s V98-22 s V98-23 s V98-24 s V98-25 r L-1	<5 <5 16 7 <5	



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25/09/98	Certificate of A	nalysis	Page
Larry	Carlyle		NO#05614
		Certified by	<u>A</u> P
Sample #	Au ppb	0	
r LV-1	<5		
	<0		

3



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Page 1

	Larry Carlyle	9		Certified by	A	WO#05614a
Sample #	total pulp wt gm	wt of +150 gm	Au in -150 oz/ton	Au in +150 mg	total Au oz/ton	
r L-2 r L-3	gm 273.7 303.2	gm 28.863 28.915	<pre>oz/ton <0.001 0.001</pre>	mg <0.001 <0.001	02/ton <0.001 0.001	
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CERTIFICATT OF ANALYSIS iPL _8I1039

2036 Columbia et Vancouver, B.C.

INTERNATIONAL PLASMA LABORATORY LTD.

64 Samples

Canada V5Y 3E1 Phone (604) 879-7878 Fax (604) 879-7898 Out: Oct 02, 1998 1 of 2

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Sample Name	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm	Hg ppm p	Mo T1 pm ppm	Bi ppm	Cd ppm	Co ppm	Ni ppm	Ba V ppm ppm	l Cr i ppm	V ppm	Mn ppm	La ppm	Sr ppm	Zr ppm	Sc ppm	Ti X	A1 *	Ca ¥	Fe X	Mg X	K X	Na X	P ¥	
C98 - 1 C98 - 2 C98 - 3 C98 - 4 C98 - 5	₽ < ₽ < ₽ < ₽ 0.1 ₽ <	21 19 21 17 15	5 8 8 7 11	28 31 45 32 38	29 31 56 34 37	~ ~ ~ ~ ~	~ ~ ~ ~ ~	2 < 2 < 1 < 1 < 1 <	~ ~ ~ ~ ~	4.6 5.0 6.9 4.6 5.3	8 11 14 9 14	22 27 44 23 23	121 64 80 91 139	26 31 53 26 30	43 43 56 43 51	163 157 226 167 365	9 10 12 13 10	13 13 11 11 19	2 3 2 1 3	2 0 2 0 3 0 2 0 2 0	.05 .06 .05 .04 .07	0.88 0.89 1.37 0.97 1.17	0.20 0.21 0.17 0.15 0.25	2.25 2.44 3.45 2.40 2.57	0.37 0.41 0.72 0.38 0.36	0.06 0.06 0.09 0.06 0.07	< < < < < <	0.02 0.03 0.04 0.03 0.04	
C98 - 6 C98 - 7 C98 - 8 C98 - 9 C98 - 10	P < P 0.1 P 0.1 P 0.1 P <	10 16 15 32 12	8 5 8 4	35 41 33 48 30	30 51 31 48 30	~ ~ ~ ~ ~	< < < < <	1 < < < 2 < 1 <	VVVV	4.6 5.9 4.5 6.3 3.7	7 12 8 14 9	12 19 17 35 20	93 150 105 92 129	27 30 28 37 27	52 63 45 58 40	238 262 150 257 255	9 9 9 11 8	13 21 15 31 16	1 1 5 1	2 0 3 0 2 0 3 0 2 0	.06 .09 .05 .09 .07	1.01 1.52 0.94 1.42 0.90	0.17 0.31 0.19 0.34 0.24	2.36 2.90 2.23 3.10 1.99	0.32 0.64 0.39 0.65 0.37	0.05 0.09 0.06 0.11 0.09	< < < 0.01	0.06 0.05 0.04 0.04 0.04	
C98 - 11 C98 - 12 C98 - 13 C98 - 14 C98 - 15	P 0.1 P < P 0.3 P < P <	9 22 77 11 35	5 < 18 3 8	28 31 125 90 58	27 41 98 26 58	~ ~ ~ ~ ~	~ ~ ~ ~ ~	1 < 2 < 2 < 1 < 1 <	VVVV	3.1 3.4 10.2 3.8 6.4	6 8 24 8 15	15 22 69 14 45	115 135 348 147 120	19 21 71 21 43	32 30 65 41 47	186 216 1012 826 433	6 7 28 11 24	13 15 41 17 27	1 3 7 1 3	1 0 2 0 7 0 2 0 4 0	.06 .05 .05 .05 .05	0.79 0.95 2.55 0.92 1.36	0.22 0.29 1.11 0.23 0.55	1.52 1.74 4.58 1.80 3.16	0.26 0.40 1.19 0.21 0.78	0.06 0.15 0.41 0.05 0.18	0.02 0.03 < 0.01 <	0.03 0.04 0.10 0.05 0.11	
C98 - 16 C98 - 17 C98 - 18 S98 - 1 S98 - 2	P 0.2 P < P < P <	81 12 46 18 28	8 6 13 13 4	63 24 54 36 30	62 34 62 48 40	< < < < <	~ ~ ~ ~ ~	1 < 2 < 2 < 1 < 2 <	V V V V	6.7 3.4 6.9 5.0 4.3	14 8 17 11 7	50 20 52 30 20	245 75 123 129 128	40 21 45 36 22	45 34 47 46 38	687 170 479 331 161	19 6 25 13 15	44 14 20 16 16	2 1 6 3 1	4 0 1 0 5 0 3 0 2 0	.04 .06 .06 .06 .02	1.45 0.85 1.41 1.25 1.18	1.41 0.21 0.40 0.32 0.29	2.92 1.76 3.40 2.55 2.13	0.74 0.33 0.69 0.54 0.38	0.20 0.09 0.22 0.10 0.08	0.01 0.02 < 0.01	0.13 0.01 0.06 0.02 0.02	
S98 - 3 S98 - 4 S98 - 5 S98 - 6 S98 - 7	P < P < P 0.2 P <	10 16 21 52 23	4 6 9 14 11	30 31 39 63 42	34 35 34 59 46	~ ~ ~ ~ ~	< < < < <	1 < 1 < 1 < 2 <	~ ~ ~ ~ ~	3.5 3.8 3.9 7.4 5.9	6 8 9 20 14	12 13 27 56 40	156 238 169 320 139	19 25 28 57 46	36 39 35 54 53	256 408 289 801 335	6 10 12 25 16	16 17 23 32 17	1 1 2 2	2 0 2 0 2 0 4 0 3 0	.04 .04 .05 .07 .08	0.89 1.06 0.93 1.60 1.43	0.42 0.32 0.43 0.92 0.33	1.67 1.97 1.99 3.56 2.87	0.29 0.38 0.44 0.94 0.68	0.06 0.06 0.08 0.18 0.14	0.01 0.01 0.02 < 0.01	0.03 0.04 0.05 0.08 0.03	
S98 - 8 S98 - 9 S98 - 10 S98 - 11 S98 - 12	P < P < P < P 0.1 P 0.2	29 33 39 40 51	10 14 13 10 15	49 56 52 57 84	47 64 48 57 67	~ ~ ~ ~ ~	v v v v v	1 < 2 < 1 < 2 < 3 <	~ ~ ~ ~ ~	5.0 7.4 6.8 6.4 10.5	14 18 18 16 32	35 54 54 51 142	254 199 79 187 270	37 61 57 49 138	42 58 50 38 86	884 379 300 466 1041	14 21 23 19 21	35 20 14 136 34	1 1 7 4 3	2 0 4 0 4 0 3 0 5 0	.06 .05 .07 .04 .06	1.21 1.75 1.41 1.34 1.95	0.69 0.28 0.21 5.95 0.71	2.45 3.61 3.47 2.84 4.96	0.56 0.88 0.80 1.15 1.58	0.17 0.14 0.22 0.14 0.41	0.01 < < < <	0.11 0.05 0.02 0.09 0.10	
S98 - 13 S98 - 14 S98 - 15 S98 - 15 S98 - 16 V98 - 1	P 0.2 P < P 0.2 P 0.3 P <	47 48 91 108 18	14 13 9 7 4	68 58 72 73 33	59 52 66 74 22	~ ~ ~ ~ ~	~ ~ ~ ~ ~	2 < 1 < 2 < 1 <	~~~~~	7.6 7.0 9.5 12.1 3.1	20 19 25 35 8	62 66 104 175 16	209 154 157 249 140	56 62 107 228 19	51 47 71 103 34	780 650 1130 1516 683	23 20 28 19 16	39 27 41 52 20	3 3 3 1	30 30 30 60 10	.03 .03 .04 .04 .04	1.50 1.37 1.94 2.16 0.77	0.79 0.94 1.13 1.43 0.41	3.55 3.48 4.33 5.62 1.64	0.88 0.86 1.37 1.96 0.29	0.25 0.18 0.18 0.23 0.14	< < < 0.02	0.14 0.08 0.20 0.17 0.03	
V98 - 2 V98 - 3 V98 - 4 V98 - 5	P < P 0.1 P 0.1 P 0.1 P 0.1	14 13 36 14	3 8 7 4	21 23 34 26	26 43 53 33	~ ~ ~ ~	< < 3 <	1 < 2 < 1 < 1 <	V V V V V V	3.5 3.8 5.4 3.5	10 8 12 8	27 19 40 22	70 100 186 71	40 29 39 26	37 40 49 35	147 183 464 172	7 8 14 7	18 20 20 16	2 2 4 2	2 0 2 0 4 0 2 0	.09 .08 .07 .06	0.79 1.00 1.51 0.88	0.37 0.43 0.43 0.26	1.89 1.92 2.65 1.79	0.44 0.37 0.49 0.40	0.08 0.11 0.18 0.07	0.01 0.01 0.01 0.01	0.01 0.01 0.02 0.03	
Min Limit Max Reported* Method ————————————————————————————————————	0.1 99.9 ICP s=Insufficio	1 20000 2 ICP ent Samr	2 20000 2 ICP ble De	1 20000 ICP I=Delay	5 9999 ICP Max	5 999 9 ICP (=No	3 9999 9 ICP I Estimate	1 10 99 999 CP ICP e Rec=	2 9999 ICP ReChec	0.1 99.9 ICP	1 9999 ICP =x1000	1 9999 ICP %=I	2 9999 999 ICP ICF Estimate %	1 9999 ICP NS=N	2 9999 ICP Io Sarr	1 9999 ICP pleP=F	2 99999 9 ICP Vulp	1 9999 ICP	1 9999 ICP	10 99991 ICP	.01 .00 ICP	0.01 9.99 ICP	0.01 9.99 ICP	0.01 9.99 ICP	0.01 9.99 ICP	0.01 9.99 ICP	0.01 5.00 ICP	0.01 5.00 ICP	



CERTIFICATE OF ANALYSIS iPl JI1039

2036 Columbia Street Vancouver, B.C Canada V5Y 3E

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INTERNATIONAL PLASMA LABORATORY LTD.

Ag ppm

Project: W.O. 5614

Sample Name

L . ۱L . IL. ۱L۷ . M .

Client : Northern Analytical Laboratories

Cu

ppm

54 S 64	am =Pulj	ples	ł								[1	.03913	:19:3	5:891	.00298]	Out: In :	Oct 0 Sep 2	Phi Fax 2, 1998 5, 1998	one (60 : (60	4) 87 4) 87 Pa Se	9-7878 9-7898 Ige ection	2 of 1 of	2 1
Hg ppm	Mo ppm	T1 ppm	Bi ppm	Cd ppm	Co ppm	Ni ppm	Ba ppm	W ppm	Cr ppm	V ppm	Mn ppm	La ppm	Sr ppm	Zr ppm	Sc ppm	Ti X	רא ג	Ca X	Fe X	Mg X	K X	Na X	P X	
~ ~ ~ ~ ~	1 2 1 1 2	< < < < <	V V V V V	3.3 3.1 4.5 3.9 5.6	7 6 12 10 13	19 14 53 20 33	59 72 96 79 143	~ ~ ~ ~ ~	22 23 49 34 34	32 36 40 35 43	140 120 290 195 443	5 6 8 6 14	11 14 19 16 18	2 2 2 1 3	1 2 3 2 4	0.06 0.06 0.07 0.07 0.05	0.71 0.75 0.89 0.79 1.34	0.17 0.23 0.45 0.27 0.27	1.62 1.59 2.22 1.88 2.76	0.33 0 0.31 0 0.67 0 0.50 0 0.65 0	.06 .05 .16 .09 .23	0.01 0 0.01 0 0.01 0 0.01 0 < 0	.01 .01 .03 .01 .04	
< < < < < <	1 1 2 1 3	~ ~ ~ ~ ~	~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~	4.0 3.4 5.4 3.8 4.0	9 8 14 9 10	20 16 26 26 29	82 105 53 94 105	~ ~ ~ ~ ~	22 28 38 37 35	36 40 56 36 36	282 159 285 206 279	10 8 6 13 10	14 15 13 14 16	3 2 3 2 3	2 2 2 3	0.06 0.08 0.11 0.07 0.07	0.97 0.96 1.35 0.90 0.92	0.24 0.23 0.24 0.30 0.35	2.06 1.85 2.73 1.93 2.09	0.46 0 0.43 0 0.86 0 0.45 0 0.50 0	.12 .11 .55 .13 .17	0.01 0 0.01 0 < 0 0.01 0 0.01 0	.03 .02 .01 .01 .01	
< <	1	< <	<	4.6 3.3	12 10	35 32	97 42	×	38 42	41 33	341 136	9 4	19 10	2 1	2	0.05	1.01	0.52	2.34	0.65 0 0.50 0	.15	< 0 0.01 0	.04	

V98 - 6 V98 - 7	P < <	10 4 8 6	23 23 21 26	< <	< <	1 2	<	3.3 3.1
V98 - 8 V98 - 9	P. < P. <	25 7 12 7	33 32	< <	< <	1	< _<	4.5
V98 · 10	P <	37 10	45 53	<	<	ź	< <	5.6
V98 - 11	₿ <	24 7	33 38	<	<	1	< <	4.0
V98 · 12	P <	11 9	27 26	<	<	1	<	3.4
V98 · 13	P 0.1	21 12	54 46	<	<	2	<	5.4
V98 - 14	P 0.1	16 6	27 26	<	<	1	< 2004	3.8

Pb

ppm

64 Samples

As Sb

ppm ppm

Zn

ppm

V98	- 15	Р	0.1	23	5	30	30	<	<	3	<	4.0	10	29	105 <	35	36	279	10 16	3	3 0.07 0.92 0.35 2.09 0.50 0.17 0.01 0.01
V98 V98 V98 V98 V98 V98	- 16 - 17 - 18 - 19 - 20	P P P P	~ ~ ~ ~ ~	38 13 11 22 8	5 6 10 7 5	38 23 25 34 19	33 18 22 36 20	< < < < <	~ ~ ~ ~ ~	1 < 1 2 1	< < < < < < < < < < < < < < < < < < <	4.6 3.3 3.5 4.8 3.3	12 10 11 15 8	35 32 31 55 19	97 < 42 < 70 < 60 < 65 <	38 42 46 61 28	41 33 34 40 32	341 136 261 300 144	9 19 4 10 5 14 9 15 5 16	2 1 1 2 1	$\begin{array}{cccccccccccccccccccccccccccccccccccc$
V98 V98 V98 V98 V98 V98	- 21 - 22 - 23 - 24 - 25	P P P P	0.1 < < < 0.1	30 17 19 26 67	8 8 5 9 14	40 29 29 32 69	46 27 29 33 54	< < < < <	~ ~ ~ ~ ~	1 1 1 5	~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~	5.1 4.1 3.7 4.1 7.6	13 11 10 10 19	31 28 28 28 54	123 < 85 < 79 < 46 < 119 <	32 37 29 33 49	45 37 34 34 59	449 209 334 190 645	12 22 7 17 7 16 11 16 25 30	2 1 3 5	$\begin{array}{cccccccccccccccccccccccccccccccccccc$
L L LV M	- 1 - 2 - 3 - 1 - 1	P P P P	0.1 0.2 0.1 < 0.1	42 36 22 2 1	11 6 2 11	62 58 9 28 10	54 49 < 21 <	< < < < <	~ ~ ~ ~ ~	3 2 1 1 4	~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~	7.5 3.8 2.4 1.2 0.4	16 6 8 2 <	37 15 25 4 <	89 < 111 < 18 < 4 < 10 <	103 24 16 53 11	96 31 2 9 <	793 990 155 330 147	11 72 9 167 < 49 10 44 2 102	6 15 3 7 1	$\begin{array}{cccccccccccccccccccccccccccccccccccc$

Min Limit Max Reported* Method



03/06/98

Assay Certificate

Page 1

Larry Carlyle

Certified by

WO#07978

Sample #	Au ppb				
ciot-1 cot-2 cot-3, cot-4 cot-5	5 5 7 < 5 < 5				
cot-6 cot-7 cot-8 cot-9 cot-10	7 10 6 < 5 < 5				••
cot-11 cot-12 cot-13 cot-14 mandy-1	8 7 5 5 98				
ron-1 T1-S2 T2-S1 T2-S2 T3W-S1	33 234 18 28 > 7000	@ 74 ,0m			
T3W-S2 T3W-S3 T3W-S4 T4-S1 T4-S2	254 > 7000 22 43 29	@ 71,0m @ 89.0m			· .
T4-S3 windlass 1 windlass-2 windlass-3	15 21 9 20				





12/06/98

Assay Certificate

Page 1

Larry Carl	lyle	WO# 07989
		Certified by
Sample #	Au oz/ton	0
T3W - S1 T3W - S3	0.285 0.938	(FA/AAS) @ 750 m (FA/Gravimetric) @ 89m
·		





CERTIFICA^{TT} OF ANALYSIS iPL **78E0490**

2036 Columbia et Vancouver, B.C.

Canada V5Y 3E1

	INTERNAT	TIONAL F	PLASMA LA	BORATORY	LTD.																						P	hone ax	(604) (604)	879-787	78 38	
Client Project	: Nor : W.O	ther	n Ana 178	lytica	1 Labor	ratori	es	2	29 Sa 29=	a mple Pulp	\$						[0	49012	2:04:2	1:890	152798	8]			Out: In :	May 2 May 2	7, 19 5, 19	98 98		Page Sectio	10 n 10	of 1 of 1
Sample	Name	•	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm	Hg ppm	Mo T1 ppm ppm	Bi ppm	Cd ppm	Co ppm	Ni ppm	Ba ppm	W ppm	Cr ppm	V ppm	Mn ppm	La ppm	Sr ppm	Zr ppm	Sc ppm	Ti X	A1 *	Ca ¥	Fe X	Mg X	K X	Na X	P X	
COT - COT - COT - COT - COT -	1 2 3 4 5		< < 0.1 <	1 4 9 6 7	< < 4 < 6	9 13 23 15 19	< < 8 6 4	< < < < <	< < < < <	<pre>< < < < < < < < < < <</pre>	V V V V	0.4 0.4 0.9 0.7 0.7	< 3 9 4 6	3 5 13 6 10	21 82 139 71 141	~~~~	1 6 19 10 12	7 17 34 26 24	13 250 633 87 608	< 2 5 4 4	6 6 12 10 10	< 1 < 1 <	< (< (1 (1 (1 ().01).03).04).04).04	0.12 0.36 0.72 0.57 0.55	0.04 0.07 0.16 0.12 0.14	0.27 0.82 1.80 1.13 1.31	0.02 0.09 0.25 0.14 0.16	0.01 0.02 0.04 0.03 0.04	0.01 0.01 < (0.01).02).02).03).05).04	
COT - COT - COT - COT - COT -	6 7 8 9 10		< < < < <	8 35 10 9 7	< 7 5 4 <	22 43 23 22 18	6 27 12 8 «	~ ~ ~ ~ ~	~ ~ ~ ~ ~	1 < 1 < 1 < 1 <	VVVV	0.7 1.4 1.0 0.9 0.7	7 13 9 8 6	15 36 22 19 9	76 137 134 114 99	~~~~	17 32 23 22 13	28 35 34 31 26	142 443 204 149 123	5 17 5 6 4	11 17 11 11 8	1 4 1 1 1	1 (3 (1 (1 (1 ().03).04).03).04).03	0.68 1.19 0.91 0.77 0.57	0.17 0.30 0.18 0.15 0.11	1.62 2.72 2.04 1.83 1.30	0.28 0.57 0.37 0.34 0.19	0.05 0.12 0.05 0.07 0.04	< (< (< (< (02 06 01 02 02 02	
COT - COT - COT - COT - Mandy	11 12 13 14 · 1	P. P. P. P.	< < < 0.5	11 7 1 4 16	6 3 < 36	26 26 5 7 6	11 6 < < <	~ ~ ~ ~ ~	~ ~ ~ ~ ~	1 < 1 < < < < <	V V V V	1.0 0.8 0.2 0.3 0.7	8 5 1 1 3	20 11 2 1 7	112 111 29 30 67	~ ~ ~ ~ ~	24 17 1 26	38 29 9 11 10	142 108 19 19 263	5 5 < 18	9 10 6 99	1 < < 3	1 (1 (< (1 ().03).03).01).01).01	0.91 0.80 0.14 0.14 0.27	0.13 0.14 0.04 0.03 2.24	2.06 1.47 0.34 0.42 0.92	0.37 0.27 0.04 0.02 0.09	0.04 0.03 0.02 0.01 0.02	<pre>< (</pre>).02).04).02).02).02	
Ron T1 - S T2 - S T2 - S T3W -	1 2 51 52 51	P P P P	1.9 1.9 < 94.7	432 244 32 31 73	75 274 10 15 6427	130 25 68 57 4	121 19 < 11 <	26 < < < < <	~ ~ ~ ~ ~	5 < 3 < 2 < 1 <	< < < 119	3.1 1.5 2.1 1.8 1.0	5 14 16 16 1	31 13 41 55 5	147 353 102 99 555	~ ~ ~ ~ ~	153 117 75 90 139	18 17 44 40 3	271 2621 767 720 52	 3 18 14 	4 25 67 56 17	< 7 6 9 3	1 1 4 (4 (<	<pre> </pre> .04 .04 .04 .04	0.10 0.14 1.17 1.14 0.07	0.03 0.16 2.46 2.78 0.02	3.99 2.61 3.88 3.33 1.58	0.02 0.02 1.02 1.03 0.02	0.02 0.11 0.15 0.03	< < < < <	0.04 0.07 0.19 0.11 <	
T3W - T3W - T3W - T4 - S T4 - S	S2 S3 S4 1 2		0.4 0.1m 0.7 <	23 86 13 45 35	29 15110 82 17 14	21 4 8 48 56	< < 20 15	~ ~ ~ ~ ~	~ ~ ~ ~ ~	1 < 4 < 1 < 2 < 1 <	: × 173 × × ×	1.0 2.3 0.4 1.6 1.6	7 2 3 14 17	14 6 38 38	44 225 16 118 146	~ ~ ~ ~ ~	117 138 170 112 74	6 2 25 21	240 50 85 780 1072	11 < 17 21	7 15 1 12 12	21 5 3 10 12	2 < 3 (3 (< < < 0.01).01	0.25 0.05 0.06 1.05 0.99	0.08 0.01 0.06 0.21 0.20	1.91 2.63 0.60 2.99 2.97	0.05 0.01 0.02 0.89 0.86	0.06 < 0.02 0.11 0.10	< (< 0.01 (< (< ().02 <).01).06).06	
T4 - S Windla Windla Windla	3 ss - ss - ss -	P 1 P 2 P 3 P	~ ~ ~ ~	29 47 21 43	12 16 < 5	51 83 6 52	< 21 22 16	< < < <	< < < <	2 < 2 < 1 < 4 <	V V V	1.5 2.5 0.5 1.6	14 17 3 17	39 31 12 81	91 855 99 43	~ ~ ~ ~	79 58 119 130	26 73 3 53	633 1127 157 457	17 11 < 4	14 30 3 14	8 9 2 2	2 (12 (1 6).02).01 < <	0.87 1.00 0.06 0.61	0.24 0.52 0.02 0.42	3.06 4.65 0.73 3.46	0.59 0.36 0.02 0.50	0.11 0.21 0.02 0.05	< (< (< ().06).14).01).08	
						-																									•	
L Min Lim Max Rep Method	it orted ^a	*	0.1 99.9 2	1 20000 :	2 20000 2	1 20000 JCP	5 9999 TCP	5 999 9	3 9999	1 10 999 999	2 9999 ICP	0.1 99.9	1 9999 ICP	1 9999 ICP	2 9999	5 999 TCP	1 9999 ICP	2 9999 ICP	1 9999 ICP	2 9999 ICP	1 9999 TCP	1 9999 ICP	1 (9999 1 ICP).01 L.00	0.01 9.99	0.01 9.99 ICP	0.01 9.99 ICP	0.01 9.99	0.01 9.99	0.01 5.00	0.01 5.00	


29/06/98

Assay Certificate

Page 1

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WO#07998

Larry Carlyle

		Certified by
Sample #	Au ppb	(/
L + I.V 1 RON - 2 T1 - S3 T1 - S4 T1 - S5	5 5 <5 11 8	
T1 - S6 T1 - S7 T1 - S8 T3E - S1 T3E - S2	14 256 19 . 7 12	
T3E - S3 T3W T3W CG - 1 CG - 2	<5 4918 85 9 25	
CG - 3 CT - 1 LK98 - 1 LK98 - 2 LK98 - 3	9 6 <5 <5 <5	
LK98 - 4 LK98 - 5 LK98 - 6 LK98 - 7 LK98 - 8	<5 <5 <5 <5 <5	
LK98 - 9 LK98 - 10 LK98 - 11 LK98 - 12 LK98 - 13	<5 31 6 <5 <5	



29/06/98

Assay Certificate

Page 2

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WO#07998

Larry Carlyle

		Certified by
Sample #	Au ppb	0
LK98 - 14 LK98 - 15 LK98 - 16 LK98 - 17 LK98 - 18	5 <5 9 <5 <5	
LK98 - 19 LK98 - 20 SG - 1 SG - 2 SG - 3	9 <5 <5 5 <5	
SG - 4 SG - 5 SG - 6 SG - 7 SG - 8	<5 <5 <5 7	
SG - 9 SGE - 1 SGE - 2 SGE - 3 SGE - 4	6 <5 8 <5 6	
SGE - 5 SGE - 6 SGE - 7 SGE - 8	<5 5 <5 6	



J

CERTIFICATE OF ANALYSIS iP\ 8F0575

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2036 Columbia Street Vancouver, B.C Canada V5Y 3E . Phone (604) 879-7878 Fax (604) 879-7898

Client : North Project: WO#79	hern Ana 998	lytica	Labor	ratori	es		54 S 54=	ample Pulp	s					_		[0	57513:31:	57:890	62598	8]	Out: In :	Jun 2 Jun 1	5, 19 8, 19	98 98	()	Page Sect	1 ion 1	of 2 of 1
Sample Name	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm	Hg ppm	Mo Ti ppm ppr	l Bi n ppm	Cd ppm	Co ppm	Ni ppm	Ba W ppm ppm	Cr ppm	V ppm	Mn ppm	La Sr ppm ppm	Zr ppm	Sc ppm	Ti X	A1 X	Ca X	Fe X	Mg X		K Na K X	р Х	
CG-1 CG-2 CG-3 CT-1 L+IV-1	P 0.3 P 0.2 P 0.4 P 0.3 P 0.4	14 27 29 28 160	10 6 5 9 10	30 46 43 37 59	16 22 13 23 <	< < < < <	< < < < <	2 2 1 2 4		1.6 1.5 0.7 1.2 1.6	9 12 4 11 13	20 32 21 33 20	98 < 121 <	26 34 13 32 108	59 47 15 40 31	107 250 259 263 753	7 13 10 11 7 109 11 15 16 16	1 2 2 1 1	2 3 1 3 5	0.07 0.06 0.02 0.06 <	1.11 1.36 0.53 1.14 0.56	0.17 0.16 3.26 0.25 0.50	2.60 2.91 0.82 2.37 2.66	0.33 0.63 0.29 0.46 0.18	0.00	5 < 3 < 3 0.02 7 < L <	0.01 0.04 0.08 0.02 0.03	
LK98- 1 LK98- 2 LK98- 3 LK98- 4 LK98- 5	P 0.2 P 0.1 P 0.2 P 0.2 P 0.2	4 8 10 7	<	10 8 12 29 23	< 5 < 10 10	~ ~ ~ ~ ~	< < < < <	<pre>< 4 1 < 1 4 1 4 1 4 1 4 1 4 1 4 1 4 1 4 1</pre>	~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~	0.4 0.5 0.5 1.0 0.9	3 4 4 7 6	3 4 8 12 11	32 < 66 < 57 < 129 < 77 <	2 4 8 18 16	16 17 22 39 39	75 323 198 325 189	< 7 3 16 6 11 7 16 8 10	< < < < <	< < 1 1 1	0.04 0.03 0.03 0.04 0.06	0.22 0.33 0.44 0.89 0.74	0.05 0.23 0.14 0.21 0.13	0.64 0.68 1.02 1.75 1.42	0.06 0.08 0.13 0.26 0.26	0.03	3 0.03 3 0.03 5 0.02 5 0.01 5 0.01	0.02 0.03 0.02 0.03 0.03	
LK98- 6 LK98- 7 LK98- 8 LK98- 9 LK98-10	P 0.2 P < P < P 0.2 P 0.3	5 2 4 24 19	5 < 2 6	19 - 7 12 14 18	9 < 7 5	~ ~ ~ ~ ~	< < < < <	1 · · · · · · · · · · · · · · · · · · ·		0.8 0.4 0.5 0.4 0.8	7 2 3 3 7	8 4 6 7 14	103 < 18 < 39 < 127 < 59 <	13 3 5 3 12	35 15 14 9 25	438 32 113 346 217	5 11 < 5 3 8 5 63 9 17	< < 1 1	1 < < 2	0.04 0.02 0.01 0.02 0.03	0.83 0.12 0.22 0.55 0.63	0.13 0.03 0.09 1.67 0.34	1.41 0.49 0.61 0.49 1.71	0.21 0.04 0.07 0.10 0.18	0.04 0.03 0.03 0.03 0.03	4 0.01 3 0.03 3 0.03 3 0.03 6 0.01	0.02 0.01 0.02 0.06 0.01	
LK98-11 LK98-12 LK98-13 LK98-14 LK98-15	P 0.3 P 0.2 P 0.3 P 0.2 P 0.2	18 22 22 9 19	11 8 < 4 7	27 36 33 19 36	12 13 < 13	< < < < <	< < < < <	2 1 4 1 2	~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~	1.3 1.4 1.0 0.7 1.5	8 12 3 5 9	15 24 7 14 25	60 < 72 < 113 < 57 < 88 <	23 24 3 17 28	45 34 12 24 49	370 688 188 143 203	22 9 25 14 3 80 8 9 11 12	< 1 1 < 1	2 2 < 1 2	0.04 0.03 0.02 0.03 0.04	1.17 1.20 0.39 0.51 1.25	0.10 0.19 1.55 0.09 0.14	2.30 2.62 0.59 1.33 2.89	0.32 0.44 0.13 0.22 0.41	0.0	7 < 9 < 4 0.03 5 0.02 7 <	0.02 0.02 0.04 0.02 0.02 0.02	·
LK98-16 LK98-17 LK98-18 LK98-19 LK98-20	P 0.3 P 0.3 P 0.3 P 0.3 P 0.3 P 0.2	4 8 3 13 4	< 14 < 5 7	10 25 12 34 13	< 16 < 10 <	~ ~ ~ ~ ~	< < < < <	1 1 1 1 1	~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~	0.5 0.9 0.3 1.1 0.5	3 5 3 10 3	7 13 3 29 8	40 < 84 < 81 < 41 < 34 <	9 19 2 28 8	21 33 13 29 22	89 132 507 262 56	5 8 8 10 4 22 15 15 10 7	v v v v v	< 1 < 2 1	0.04 0.04 0.03 0.04 0.05	0.36 0.81 0.33 0.79 0.35	0.10 0.14 0.38 0.27 0.08	0.79 1.52 0.55 2.22 0.70	0.12 0.30 0.10 0.53 0.17	0.0	5 0.02 5 0.01 3 0.04 5 < 4 0.02	0.02 0.02 0.05 0.08 0.01	
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Project: WO#7998

Sample Name

SGE-5

SGE-6

SGE-7

SGE-8

TI-S3

CERTIFICA OF ANALYSIS iPL 98F0575

2036 Columbia .et Vancouver, B.C. Canada V5Y 3E1

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www.com

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91

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42

8

ppm

Min Limit Max Reported* Method

APPENDIX C INVOICES SUPPORTING STATEMENT OF COSTS

LIVINGSTONE PLACER LTD.

Bulldozer Trenching (96.5 hrs @ \$185./hr)		\$ 17,852.50
Air Charters		\$ 1,059.70
Room & Board (53 person/days @ \$35/day)		\$ 1,855.00
Miscellaneous Fuels & Oil		\$ 200.00
ATV Rental (2 wks. @ \$125/wk)		\$ 250.00
	TOTAL	\$ 21,217.20

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LIVINGSTONE PLACER LTD. BULLDOZER INVOICE

Bulldozer: Terex D 800 Series (D - 9 equivalent) Equipped with U-Blade and Rippers

Bulldozer was utilized from May 1 - 8 for at least 10 hrs/day for road clearing, trench construction, as well as mob and demob.

Bulldozer use 80 hours @ \$185./hr \$ 14,800.00

LIVINGSTONE PLACER LTD. BULLDOZER INVOICE

Bulldozer: Terex D 800 Series (D - 9 equivalent) Equipped with U-Blade and Rippers

Bulldozer was utilized from October 23 - 24 for trench construction, as well as mob and demob.

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Bulldozer use 16.5 hours @ \$185./hr \$3,052.50

CARLYLE INVOICE

LIVINGSTONE CREEK PROJECT

Geologist Field Work (24 days @ \$300/day)		\$	7,200.00
Assaying		\$	3,828.56
Report Writing		\$	1,500.00
Air Charters		\$	758.70
Field Supplies (Flagging, bags, hip chain twine, etc.)		\$	200.00
Office Supplies (Photocopying, paper, etc.)		\$	193.08
	TOTAL	\$1	3,680.34

CARLYLE FIELD WORK INVOICE

Carlyle Wages (May 4 - 8 @ \$300./day)		\$ 1,500.00
Carlyle Wages (June 6 - 10 @ \$300./day)		\$ 1,500.00
	TOTAL	\$ 3,000.00

CARLYLE FIELD WORK INVOICE

 Carlyle Wages (October 23 - 25 @ \$300./day)
 \$ 900.00

 TOTAL
 \$ 900.00

CARLYLE FIELD WORK INVOICE

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Carlyle Wages (September 14 - 19 @ \$300./day)	\$ 1,800.00
Carlyle Wages (September 28 - October 2 @ \$300./day)	\$ 1,500.00
TOTAL	\$ 3,300.00



Invoice Date: 03/06/98

Larry Carlyle

To:

WO# 07978

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QTY	DESCRIPTION	UNIT PRICE	AMOUNT
7	Sample Preparation: Sample Drying	2.50	17.50
14	Rock	2.00	28.00 75.00
	Analyses:		
29	Au + 30	16.00	464.00
	PAID CHQ 296.		
	Subtotal		584.50
	GST @7% (R 121285662)	40.92
ingstone	Assays		
v	J Total due on receipt of invo	pice	\$625.42
	2% per month charged on	overdue accou	ints
	LESS 15 CON	nons (172.50
·	TOTAL		452.92

105 Copper Road, Whitehorse, YT, Y1A 2Z7 Ph: (403) 668-4968 Fax: (403) 668-4890



To: Larry Carlyle Invoice Date: 12/06/98

WO# 07989

\$24.61

QTY	DESCRIPTION	UNIT PRICE	AMOUNT
1	Analyses: Au 1AT FA/AAS Au 1AT FA/Gravimetric	11.00 12.00	11.00 12.00
L	Subtotal	<u> </u>	23.00
in a track	GST @7% (R 121285662)		1.61

Livingstone Assays

Total due on receipt of invoice

2% per month charged on overdue accounts

PAD CASH

105 Copper Road, Whitehorse, YT, Y1A 2Z7 Ph: (403) 668-4968 Fax: (403) 668-4890



Invoice Date: 18/06/98

Larry Carlyle

To:

WO# 07998

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QTY	DES	CRIPTION		JNIT PRICE	AMOUNT
13 41	Sample Preparation: Rock/D.C. Sample Pre Soil/Sediment Sample	paration Preparation		5.00 2.00	65.00 82.00
54	Analyses: Au + 30			16.00	864.00
<u></u> _		Subtotal	I.		1011.00
ingsto	ne Assays	GST @7% (R 1212	285662)		70.77
	v	Total due on receip 2% per month charg	t of invoid ged on ov	ce verdue accou	\$1,081.77 ints
	13 ASSAY COUPONS	(\$273.00)		NET	\$ 808.77

105 Copper Road, Whitehorse, YT, Y1A 2Z7 Ph: (403) 668-4968 Fax: (403) 668-4890



105 Copper Road Whitehorse, Yukon Y1A 2Z7 Ph: (867) 668-4968 Fax: (867) 668-4890 E-mail: NAL@hypertech.yk.ca

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Invoice for Analytical Services

To: Larry Carlyle Invoice Date: 25/09/98

WO# 05614

QTY	DESCRIPTION	UNIT PRICE	AMOUNT
5 59	Sample Preparation: Rock/D.C. Sample Preparation Soil/Sediment Sample Preparation	5.00 2.00	25.00 118.00
62 2	Analyses: Au + 30 Au Metallics Fire Assay + ICP-30	16.00 37.25	992.00 74.50
	Éano Subtotal		1209.50
Living	GST @7% (R 121) #040 22 Assay Coupons	285662)	84.67 (\$226.75)
YAID 2	Total due on receip	t of invoice	\$1,067.42

2% per month charged on overdue accounts



Invoice Date: 13/11/98

Larry Carlyle

To:

WO# 05625

\$1,029.34

QTY	DESCRIPTION	UNIT PRICE	AMOUNT
42 32	Sample Preparation: Rock/D.C. Sample Preparation Sample Drying	5.00 2.50	210.00 80.00
42	Analyses: Au + 30	16.00	672.00
	Pairs CHIQ O W-D.	52.	
	Subtotal		962.00
	GST @7% (R 121285662)	1	67.34

Livingstone Assays

Total due on receipt of invoice

2% per month charged on overdue accounts

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BIG SALN 668-2 P.O. BO Whitehorse, Yu AC. <u>(E55N/A</u>	10N 1608 x 6001 Jkon Y1A <u>203</u> J	AIR 1517	DATE	CHARTER TICKET Nº 2388 June 6 98	BIG SALN 668-4 P.O. BO Whitehorse, Yu AC <u>CESSINA</u>	10N 1608 x 6001 11kon Y1/ こしん	AIR (PA ## 2225	⁶ CHARTER Nº Oct 2	TICKE 211	T 0 8
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PacBlue Digital Reprographics Inc.

1595 West oth Avenue Vancouver BC V6J 1R1 CANADA

Telephone: (604) 714-3288 Fax: (604) 714-3289

GST #: 13281 2538

Bill To:

CASH SALES 1595 West 6th Ave. Vancouver, B.C. V6J 1R1

invoice #:

Invoice Date:

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Ship To:

Larry W. Carlyle 74 Tamarack Drive Whitehorse Yukon Y1A 4Y6

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Livingstone Claim Area Topo map blown up from 1:50,000 to 1:10,000 scale

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Docket No.



LIVINGSTONE CREEK

Max Fuerstner, L. Carlyle

Claims: CAM 1 - 142 NTS: 105 E/8 Coordinates: 61⁰ 19' N; 134⁰ 17' W

Area: Whitehorse

Access: Fixed-wing aircraft to Livingstone airstrip; winter road from Lake Laberge

98-004

1

Commodities: Au

INTRODUCTION:

The Livingstone Creek placer camp has been worked for about 100 years. Its production is estimated to be approximately 70,000 ounces (30,000 oz. from lower Livingstone Creek). Much of this production has been large nuggets (up to 39 oz.) having a purity of 860 - 895 fineness. It was believed that nuggets of this size and quality could not have migrated far from their source. It was also surprising that so little hard rock exploration had been done in an area with such an extensive placer mining history.

HISTORY:

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Prospecting during 1996 revealed prominent white bull quartz veins having widths from 4 inches up to 4 feet, but most commonly having a width of 1 to 2 feet, located within faults having a strike about 320^o Az (almost perpendicular to the placer streams with steep west dips). Faults were evident by the existence of sharp depressions which could be traced for significant distances across the hillsides. Further evidence of the faults was increased alteration of the biotite schist to chlorite then to sericite as the faults were approached. Strong shearing and light to dark brown iron oxide occurs within the faults. Very little mineralization other than trace oxidized pyrite was seen in any of the quartz exposed on surface. The best gold, arsenic, and copper values seemed to come from samples which had strong iron and manganese staining and/or the presence of graphitic material in them.

Strong pyrite, galena, as well as copper and silver sulphides were not present in the quartz on the ridges and gullies. This mineralization was only seen in vein quartz from the Horseshoe Adit (See 1997 Work Areas on Cam Claims) and vein quartz from the placer workings on Livingstone Creek. It was concluded that the mineralization had been concentrated by some means below a certain elevation; possibly a paleo-watertable.

I confirmed to Larry Carlyle (50% owner of the claims) that I believed the gold is coming from these structures. I have placer mined on Livingstone, Lake, and Cottoneva Creeks for the past 25 years. As I mine upstream, I encounter the greatest amount of gold at these structures and less gold between structures. It is my belief that the gold from one structure migrates downstream only as far as the next structure. Most of the gold is found in the decomposed bedrock rather than in the overlying gravels.

A study of air photos showed the structures as strong lineations running parallel the Big Salmon and Teslin Faults. The faults are known to contain gold-bearing quartz-carbonate lenses and boudins. The lineations are stronger and closer together near the bottom of the placer creeks where there is less overburden. This fact could make open pit mining along the faults economic.

Claim staking started on April 30, 1997. The first 126 claims were recorded on May 16, 1997. An additional 16 claims were recorded on July 22, 1997.

GEOLOGY:

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The geology and the placer gold deposits of the Livingstone Creek area were first described by McConnell in 1901. Regional geological mapping was carried out by Cockfield, Lees, and Bostock between 1929 and 1934.

The regional geology was reinterpreted by Tempelman-Kluit in 1978-1979. This interpretation identified the Big Salmon Fault down which the South Big Salmon River flows and into which the placer creeks drain. Tempelman-Kluit, during this mapping, was the first to identify the Teslin Suture (4 - 6 miles west of the Livingstone camp) as the ancient western margin of North America (GSC Open File 1101). The rocks west of the Teslin Suture were pressed against and over the original North America during the Early Cretaceous. This action caused the rocks east of the Big Salmon Fault to be raised in reverse faulted thrust blocks (See Drawing from Tempelman-Kluit's 1979 Report). It is probable that this process caused the gold mineralization in the faults.

During the 1997 Geoscience Forum, a talk was given by Martin de Keijzer, a graduate student at the University of New Brunswick. His talk described some of the conclusions he has arrived at concerning the geological framework of the Teslin Zone and the Eastern Cassiar Terrace. He believes that Late Triassic or Early Jurassic Stikinia and Early Mississippian or older Yukon-Tanana Terranes were complexly folded into recumbent folds toward the east to lie unconformably over Devono-Mississippian Cassiar Platform rocks. This interpretation would imply a collision similar to that suggested by Tempelman-Kluit. He was unclear as to where the implied collision occurred but it can be assumed it was in the area of the Big Salmon Fault or the Teslin Fault as suggested by Tempelman-Kluit. The collision would have had to occur after Early Jurassic time; perhaps earlier than the Early Cretaceous suggested by Tempelman-Kluit. The folding would

have resulted in the high grade metamorphism seen in the rocks today. He believes the D'Abbadie Fault is steeply dipping like the Big Salmon Fault rather than being a thrust fault as interpreted by Tempelman-Kluit. It is interesting to note that most, if not all, of the creeks which have had placer production are between these two faults. This new interpretation does not change the probability that the gold is coming from the faults; their genesis may be all that has changed. They may be axial plane faults produced from the folding or faults formed simply parallel the Big Salmon and D'Abbadie Faults by interactions between them; instead of thrust faults as interpreted by Tempelman-Kluit.

CURRENT WORK:

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The work undertaken in 1997 consisted of some rock sampling, soil sampling and ground VLF-EM surveying. Most of these data are included as part of this submission. Soil samples were taken in undisturbed ground along the north rims of all of the creeks in the claim block since the overburden was expected to be thinner there as well as permafrost free. A significant number of samples returned sporadic gold values above an arbitrary 10 ppb. background. Soil samples having greater than 100 ppb.(0.1 g/t) values in gold were returned from the adit area of Livingstone Creek, and Summit, Lake, and Cottoneva Creeks. Ground VLF-EM surveys done along Livingstone, Summit, and Lake Creeks; as well as along the ridges between Livingstone and Summit Creeks and Summit and Lake Creeks (not included in submission); confirmed the existence of the faults seen as air photo lineations.

The best soil sample had a grade of 898 ppb.(0.9 g/t) and came from Summit Creek. The best rock sample came from a 0.5 m. quartz veinlet in Summit Creek and had a gold grade of 1446 ppb.(1.4 g/t). These samples are located in the area marked "B" and are very likely on the same fault as the Lake Creek Notch area marked "C". The Lake Creek Notch area returned a significant number of gold values over 100 ppb. and also showed a strong VLF response. The Notch area is expected to extend to the lower portions of Cottoneva Creek because it contains strong air photo lineations (See Drawing). The "oldtimers" also had a wagon road through the area connecting the two creeks.

The main area of interest on Cottoneva Creek is located approximately ½ mile upstream from where this road left the creek. The area marked "D" is located near a cabin at this site.

DISCUSSION:

The Livingstone property appears to have several similarities to the Macraes Mine in New Zealand. The similarities provided below will be followed up with exploration during 1998.

Similarities with the Macraes Mine:

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- 1. East-west compressional plate collision; green schist to amphibolite grade metamorphism with rapid uplift.
- 2. Long shear length 20 30 km.; shear width usually 3 5 m.
- 3. Early Cretaceous mineralization age; mineralization native gold and 2 3% sulphides.
- Best gold values found in areas of high graphitic content.

Statistics on Macraes Mine:

<u>Production:</u> 12 months ending June 30, 1997 128,797 oz. Au from 3 million tonnes of ore.

<u>Reserves:</u> 105.47 million tonnes @ 1.44 g/t. (Cut off 0.7 g/t.)tonnes.

PROPOSED WORK PROGRAM:

The work done during 1997 has demonstrated that mineralization does exist within the faults. The ground VLF-EM surveying has confirmed the air photo lineations on the ground. The 1998 work shall be concentrated in areas "A" through "D" (See 1997 Work Areas Map):

In area "A" around the old adit where VLF and grid soil sampling has shown a strong structure with coincident gold, arsenic and copper values (See drawings). The soil sampling and VLF grid will be extended to increase the size of the anomalies. The grid in this area has lines spaced at 50 metres in the north-south direction and samples separated by 20 metres. The grid will be extended 150 metres to the north. The sampling may be followed by bulldozer trenching to expose mineralization for chip sampling.

Additional claim staking may be necessary on the eastern side of the present claim block to protect the extension of structures extending from this area toward Summit Creek.

Area "B" has already provided the best gold values obtained during the 1997 program. Additional soil sampling and VLF surveying will be undertaken to extend the structure(s) toward Lake Creek where it is expected to join with area "C" (See drawings). Soil sampling with sample spacing of 25 metres has been effective in this area. Additional lines will be spaced at 50 metre intervals toward the north.

Area **"C"** will receive additional soil sampling and VLF surveying to extend the anomalies through the "Notch" toward Cottoneva along the "oldtimers" wagon road. Soil sampling along the north side of this creek was quite discontinuous because of the placer mining occurring in the vicinity. The line of soil samples

taken just north of the camp (approximately 100 metres north of the creek) indicated that two structures extend toward Cottoneva Creek (See drawings). The grid will have soil samples at intervals of 20 metres along lines spaced at 50 metre intervals. Should work in this area extend its high potential, backhoe or bulldozer trenching may be warranted.

Soil samples taken along the north rim of Cottoneva Creek in Area "D" returned several gold values of interest. Area "D" will receive VLF surveying to locate structures extending toward Little Violet Creek. Once this is done, additional soil sampling will be employed to locate anomalous gold assays in the direction of Little Violet Creek. The grid in this area is expected to have sample intervals of 25 metres along lines separated by 50 metres. If the structures extending between Lake and Cottoneva Creeks are as strong as anticipated, VLF and grid soil sampling may need to be extended downstream from area "D" to explore them.

The 1997 work left some puzzles to be solved. The biggest of these is; why does VLF surveying over lower Livingstone Creek (where most of the placer gold has come from) show strong structures but soil sampling over them return unexpectedly low gold values? The low gold values obtained from soil sampling done at Little Violet and Mendocina Creeks also are unexpected. Additional prospecting, rock and soil sampling will needed in these areas to find explanations.

Final Submission:

At the completion of the 1998 work program, a final report will be prepared describing the work done, location of samples taken with results, conclusions, recommendations, references, statement of costs, and statement of gualifications. The report will contain appendices of assay certificates and invoices supporting statement of costs.

RECLAMATION:

Any trenching or other environmental disturbance resulting from work on the property will be rehabilitated as required by the new Mining Land Use Regulations when they are passed by parliament and become law. Such rehabilitation will definitely form part of the assessment credits applied for on the properties.




TRACINGS OF

AIR PHOTO LINEATIONS

SCALE: APPROX. 1: 40,000

28°

Martin Creek

Air Photo Lineations







<u>Comparison of</u> <u>Livingstone Property</u> <u>with</u> <u>Macraes Mine, New Zealand</u>

<u>Macraes</u>

Livingstone

Tectonic Setting	Compressional	Compressional
Direction	East-West	West to East
Fault Timing	Late Tertiary	Early Cretaceous
Mineralization Age	Early Cretaceous	Early Cretaceous
Mineralization	Native Au or	Native Au
	Free Milling Au	??
	2 - 3% sulphides	Low sulphides
Mineralization P-T	325-375°C, 3+/-1kbars	Unknown
,	Rapid Uplift	Probable
Shear Length	25 km.	20 - 30 km.
Shear Dip	10-26° NE	26-75° SW
Shear Width	Few cm. to 125 m.	Few cm. to 15 m.
	Usually 5 m.	Usually 3 - 5 m.
Metamorphic Grade	Greenschist	Greenschist
	Amphibolite	Amphibolite

Best gold values found in areas of high graphitic content at both sites.

Tempelman-Kluit Mapping (O.F. 1101)

- CP_{AV} Carboniferous and/or Permian dark green, fine-grained amphibolite.
- CP_{Ag} Carboniferous and/or Permian dioritic augen amphibole gneiss.
- PM_N Paleozoic or Mesozoic pale green, strongly foliated muscovite-quartz schist.







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LIVINGSTONE CREEK



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DETAILED BUDGET

LIVINGSTONE CREEK APPLICATION

OLD ADIT

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35 soil samples @ \$19.30/ea	\$ 675.50
5 rock samples @ \$21.00/ea	\$ 105.00
Cat trenching (D9 equiv.) 20 hrs. @ \$180/hr (wet)	\$3,600.00
LIVINGSTONE CREEK	
75 soil samples @ \$19.30/ea	\$1,447.50
10 rock samples @ \$21.00/ea	\$ 210.00
SUMMIT CREEK	
60 soil samples @ \$19.30/ea	\$1,158.00
10 rock samples @ \$21.00/ea	\$ 210.00
1 day VLF surveying @ \$100/day	\$ 100.00
LAKE CREEK	
45 soil samples @ \$19.30/ea	\$ 868.50
10 rock samples @ \$21.00/ea	\$ 210.00
2 days VLF surveying @ \$100/day	\$ 200.00
20 hrs. cat trenching @ \$180/hr (wet)	\$3,600.00
COTTONEVA CREEK	
45 soil samples @ \$19.30/ea	\$ 868.50
10 rock samples @ \$21.00/ea	\$ 210.00
2 days VLF surveying @ \$100/day	\$ 200.00
LITTLE VIOLET AND MENDOCINA CREEKS	
50 soil samples @ \$19.30/ea	\$ 965.00
10 rock samples @ \$21.00/ea	\$ 210.00
2 days VLF surveying @ \$100/day	\$ 200.00
MISCELLANEOUS	
50 soil samples @ \$19.30/ea	\$ 965.00
10 rock samples @ \$21.00/ea	\$ 210.00
10 hrs. cat trenching @ \$180/hr	\$1,800.00