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	GEOLOGICAL AND GEOCHEMICA	JL		
	REPORT ON THE			
1	MOUNT GRAVES PROPERTY			
	(GRAVES 1 & 2 CLAIMS)			
	TOODOGGONE RIVER AREA,			
	OMINECA MINING DIVISION, B.	.c.		
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SSESSMEN	T REPORT 17326 MINING DIVISION: Omineca
PROPERTY:	Graves
LOCATION:	LAT 57 22 40 LONG 126 58 19 UTM 09 6361045 621942
-	NTS 094E07W
CLAIM(S):	Graves 1-2
)PERATOR (S): Blue Emerald
UTHOR(S)	: Lyman, D.A.
REPORT YE	AR: 1988, 79 Pages
COMMODITI	ES
SEARCHED	FOR: Gold, Silver, Copper, Lead, Zinc
-JEOLOGICA	L
SUMMARY:	The claims are underlain by Lower Jurassic Hazelton Group
	volcanic and volcaniclastic rocks, including welded andesite and
	pumice lapilli breccia, andesite flows and minor intrusives.
-	Propylitic alteration is pervasive. Mineralization, found in two
	zones (GWP and Yellow Rose), includes gold, silver, lead and zinc and
	is associated with fault-related guartz veining.
_VORK	
DONE :	Geological, Geochemical
	GEOL 500.0 ha
	Map(s) - 1; $Scale(s) - 1:5000, 1:1250$
-	ROCK 57 sample(s) :ME
	SILT 10 sample(s) :ME
	SOIL 116 sample(s) :ME
RELATED	
REPORTS :	10050.13458.14824
MINFILE:	094E 087

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Table 1. Production and Reserves, Toodoggone Gold Camp

1.0 SUMMARY

The Mount Graves property is comprised of the Graves 1 and Graves 2 claims totalling 35 units and is controlled under option agreement by Blue Emerald Resources Inc., Vancouver, B.C. The property is located in the Toodoggone River area of the Omineca Mining Division, 280 km north of Smithers, and is accessible by helicopter from Sturdee Airstrip. The Toodoggone River area is known to host epithermal gold deposits in Upper Triassic and Jurassic volcanics, and four properties, each having reserves over 100,000 tonnes, have been discovered to date. The Baker Mine, one of these properties, has produced 77,500 tonnes averaging 15.0 g/tonne Au and 297.8 g/tonne Ag.

Previous mineral exporation on the property over four field seasons had outlined a favourable geologic environment for epithermal gold deposits. Two mineralized areas, the GWP Zone and the Yellow Rose zone were trenched and sampled yielding values up to 11,000 ppb Au and 149 oz/ton Ag on the GWP and up to 0.306 oz/ton Au on the Yellow Rose.

Work performed during the 1987 program has confirmed the GWP Zone as the most promising prospect on the property. Results of trenching and mapping indicate probable extensions to the southwest and northeast. Rock chip sampling on the GWP Zone yielded values up to 3,000 ppb Au, 503 ppm Ag, 14,493 ppm Pb and 33,791 ppm Zn. Additional trenching and rock sampling was done on the Yellow Rose Zone with indications from mapping and soil sampling that similar mineralization may occur on a southeasterly extension. Four other mineralized areas were investigated. Additional information has been added to the geologic and structural framework of the proeprty. An exploration program is recommended which would concentrate work on the GWP Zone including core drilling, additional trenching, mapping and sampling, and a geophysical survey. More limited work is recommended for other prospects.



2.0 INTRODUCTION

At the request of the directors of Blue Emerald Resources Inc., Hi-Tec Resource Management Ltd. conducted a mineral exploration program on the Graves 1 and 2 claims in the Toodoggone River Area on August 26, 1987, and from September 1 to 18, 1987. The program was confined almost exclusively to the eastern half of the Graves 1 claim where two known gold mineralized areas, the GWP zone and the Yellow Rose Zone required further definition and possible extension along strike.

3

Other mineral showings and gossans were investigated to complete a general assessment of precious and base metal mineralization potential. Hand trenching, soil sampling on small grids, rock chip sampling and geologic mapping were employed on the established mineralized zones. Reconnaissance mapping and sampling methods were used on other showings.

A four-man crew composed of a geologist and three geological technicians conducted the work from a camp in the southernmost cirque on the property, termed the South Bowl. A statement of costs for the 1987 program is presented in Appendix II.

2.1 Property and Ownership

The Mount Graves property consists of two contiguous claims totalling 35 units, located on 23 January 1981 for Charles Kowall and still under his ownership. The claims are currently under option by Blue Emerald Resources Inc., Vancouver, B.C. Pertinent claim information is tabulated below:



Claim <u>Name</u>	Record <u>No.</u>	Record Date	Tag <u>No.</u>	No. <u>Units</u>	Area (ha.)
Graves 1	3529	Jan. 26/81	53464	15	375
Graves 2	3530	Jan. 26/81	53465	20	500
			Totals:	35	875

4

Roughly 100 ha on the eastern Graves 2 claim boundary are overstaked, and the claim boundary depicted on Figures 2 and 4 excludes the overstaked ground.

Only part of the costs of the 1987 program will be required to fulfill assessment requirements for the year ending January 26, 1988. The balance of the 1987 program will be applied to subsequent years.

2.2 Location and Access

12

Both claims are located roughly 280 km due north of Smithers, B.C., immediately southwest of Toodoggone Lake (Figures 1 and 2). The claims lie centered on approximate geographic coordinates 57°23' North latitude, 126°59' West longitude, on the western edge of NTS Map guad 94E/7W.

Immediate access is by helicopter from Sturdee Airstrip 17 km northeast to Mount Graves, which is located in the center of the property.

An alternate access is via float plane directly to Toodoggone Lake at the northeast corner of the Graves 2 claim with a short helicopter ferry to the center of the claims.

During the 1987 field season, Northern Mountain Helicopters of Smithers based a Bell 206 at Sturdee Airstrip and a Hughes 500D at the Energex camp 35 km northwest of Mount Graves. Sturdee Airstrip, at an elevation of 1200 m, has a



GENERAL LOCATION MAP GRAVES 1 & 2 CLAIMS

N.T.S.

94E

Scale: see above Figurb:

HI-TEC RESOURCE MANAGEMENT LIMITED



wide gravel surface over 1,600 m long and is suitable for landing fully-loaded cargo aircraft.

The Omineca Mining Access Road was completed to Sturdee Airstrip in mid-September 1987, opening land access from Germansen Landing 250 km northwest to the Toodoggone mining camp.

An existing access road from Sturdee Airstrip 7km northeast to the Shasta Prospect, was extended in 1987, 6 km further northeast down Jock Creek. This road construction places the nearest roadhead 9 km south from the Mt. Graves property (Figure 3).

2.3 Physiography

The main river valleys have been cut by valley glaciation to U-shaped cross sections, and the valley bottoms display hummocks and other glacial relicts. Mountainous areas are cut by V-shaped stream valleys with hanging valley remnants of recent alpine glaciation including steep cirgue walls.

Much of the property is above timberline (roughly 1700 m elevation), where grasses, sedges and dwarf willow dominate. This vegetation gives way at higher elevations to mosses and lichens, and finally bare rock. The property ranges in elevation from 1150 m at the northwest corner of the Graves 2 claim in the Toodoggone River Valley to 2070 m on the summit of Mt. Graves. Horizontally-lying evergreens and scrub pine form a dense tanglefoot on some lower portions of the mountain slopes just below timberline, but give way to normal pines below 1620 m. The broad river valleys are relatively dry and support only grasses and brush. Water sufficient for drilling is available in the northeast quadrant of the Graves 1 claim from lakes at about 1630 m elevation in the cirque east of Mt. Graves and at 1700 m elevation in the South Bowl. Drilling on the GWP zone could use snow melt from the GWP Bowl during June or early July depending on weather, thus avoiding the expense of pumping uphill from the lake.

2.4 Exploration History

The area, later to be described as the Toodoggone Gold Camp, was explored in the late 1960's for base metal deposits associated with Omineca Intrusive-driven porphyry environments, and later in the 1970's for precious metal epithermal vein-type deposits. Within a northwest-trending belt roughly 20 km wide and twice as long, four sizeable deposits have been defined, each with reserves of more than one hundred thousand tons. One of these properties, the Baker Mine, achieved production, and Cheni Gold Mines, owners of the Lawyers deposit, have scheduled the start of production for November 1988 at a rate of 550 tons per day.

A summary of these properties extracted from Schroeter & Lefebure (1987) is presented in Table 1.

The Baker Mine, Lawyers and Shasta properties lie within a 15 km arc to the south and west of Mount Graves between the Toodoggone River and Sturdee Airstrip. Discoveries of pyritic quartz veining leading to the definition of the Chappelle (Baker Mine) deposits were made by Kennco in 1969 as part of a porphyry copper exploration program (Barr, 1978). The Lawyers and Shasta properties were found in 1973 by follow-up exploration investigating northwesterlytrending regional faulting and associated alteration (Vulimiri et al., 1983, Downing & Hoffman, 1986). On the AL property, 27 km northwest of Mount Graves, Energex has TABLE 1: Production and Reserves, Toodoggone Gold Camp

Property	Operator	Tonnes (Tons)	Gra g/To (oz/1	de nne Ton)
			Au	Ag
Baker (Chapelle)	Dupont	77,500 (85,500)	15.0 (0.44)	297.8 (8.68)
	1,168,175g Au (37,55 23,084,969g Ag (710,04	7 oz) 8 oz)		
н	Multinational Res. (extension of above deposit, 1986)	55,000 (60,600)	Unspec	ified
Lawyers (3 zones)	Cheni Gold Mines	1,757,680 (1,937,000)	6.7 (0.196)	243.4 (7.10)
Shas	International Shasta	2,176,800 (2,400,000) Above reserve includes:	2.7 (0.079)	
		471,640 (520,000)	5.9 (0.172)	
AL	Energex	239,500	8.51	

operated a 6 tpd pilot mill for the last two years producing over 24,000 g Au.

The Mount Graves property was optioned by Great Western Petroleum Corp. in 1981 and reconnaissance geology and geochemistry of the Graves group of claims, plus other claims to the west and south were conducted over the next two The mineral showing now termed the GWP zone was years. discovered by this work, and returned soil values of up to 95 ppb Au and 90.5 ppm Ag, and rock values as high as 11,000 ppb Au and 7,500 ppm Ag (Caira, 1982). Based on work with Great Western, Douglas Forster completed an M.Sc. thesis that includes a regional geologic study of the southern Toodoggone River area, with a comparative economic geology analysis of eight prospects including Mount Graves. Using data from fluid inclusion analysis of mineralized rock from Mount Graves and Moosehorn Creek properties, Forster proposed a regional epithermal model that places Mount Graves at a depth favourable for hosting mixed precious and base metal deposition (Forster, 1984).

In 1984, Charles Kowall tested the gossan in the area of the Vole Peak and the head of GWP Bowl (Snowfield gossan), and possible links of the gossan to the GWP Zone, collecting soil and rock samples on cross lines and in shallow trenches. Soil samples ranged up to 280 ppb Au and 10.4 ppm Ag. Rock samples resulted in high values of 30 ppb Au and 12 ppm Ag. However, only isolated samples were anomalous and no patterns of metal zoning were detected (Kowall, 1984).

Yeager conducted assessment work in 1985 for Geostar Mining Corp. and resampled the GWP zone (Yeager and Ikona, 1986). Those rock grab samples returned gold values up to 0.170 oz/ton and silver values up to 149.21 oz/ton. Also investigated was the Yellow Rose Zone, a gossan associated with

argillic alteration and quartz veining, 550 m south of the GWP Zone. Thirty meters of trenching revealed sparse quartz veining with gold values up to 0.306 oz/ton Au in a 5 cm-wide quartz vein. Lead and zinc values over 1% were also noted.

3.0 GEOLOGY

3.1 Regional Geology

The Mount Graves property lies on the eastern margin of the Intermontane Belt, a northwesterly-trending pile of volcanic rocks and related basin sediments between the Omineca Intrusives to the east and the coast Plutonic complex to the west (Figure 3). Quartz monzonite intrusive contacts are mapped on the northwest part of the Graves 1 claim and Omineca intrusives are also present to the north and east across river valleys.

The Toodoggone River area geology has been described by Carter (1972) and Schroeter (1981). Additional work by Diakow (1984) and Panteleyev (1983 and 1984) north and south of the Toodoggone River detailed stratigraphy and structure culminating in Preliminary Map 61 (Diakow <u>et al.</u>, 1985). The oldest rocks described are Permian Asitka Group correlates mapped south of the Toodoggone River, and which consist of limestone with argillite and cherts. These rocks may actually belong to the next oldest rock unit, the Upper Triassic Takla group consisting of basaltic or andesitic flows, breccias, and tuffs.

Early to Middle Jurassic Toodoggone volcanics unconformably overlie earlier rocks and consist of well over 500 m of volcanics and volcanic-derived sediments in four divisions:

Lower volcanics -- dacite pyroclastics.



- Middle volcanics -- acidic pyroclastics that hosted explosive brecciation along faults, followed by silicification and metal deposition.
- Upper volcanic-intrusive -- crystal tuffs and quartzeye feldspar porphyries.
- Upper volcanic-sedimentary -- lake and stream sediments with interbedded tuffs.

Rocks of the Early to Middle Jurassic Hazelton Group are clearly defined south of the Toodoggone more area. Toodoggone volcanics appear to be equivalent to at least one formation of the lower Hazelton Group (Telkwa Fm.) south of the Findlay River (Panteleyev, 1984). Hazelton Group volcanics occur in the region north of the Findlay River as a northwesterly-trending belt over 50 km long by 15 km wide to the northeast of the Toodoggone volcanic belt, and have been found only in fault contact with Both Toodoggone and Hazelton Group vol-Toodoggone rocks. canics unconformably overlie Takla volcanics. In the area of Mt. Graves, Forster (1984) mapped Hazelton rocks as a series of andesite flows and lapilli breccias with related minor volcanic-derived sediments, all divided into four sub-units (Figure 3a).

Upper Cretaceous Sustut Group conglomerates and sandstones unconformably overlie earlier volcanics west of Sturdee Airstrip.

Volcanic centres elongated on northwesterly trends appear to be related to repeated normal block faulting episodes. Northeast over southwest thrust faulting may have been related to Omineca Intrusive emplacement in the Middle Jurassic. The relationship of volcanism and mineralization to northwesterly-trending faults and the regional geologic setting is summed up by Schroeter <u>et al</u>. (1986):



"Potassium-rich andesitic, subaerial pyroclastic rocks form a distinctive region within the upper part of a Mesozoic island arc-back arc complex of the Intermontane Tectonic zone. The volcanic belt is extensively block faulted (with) otherwise little disruption of stratigraphy and little metamorphism above zeolite grade. No caldera development has been recognized, ... but regional subsidence (occurred) during volcanism and local grabens developed. Northwesterly trending faults with strike lengths exceeding 20 km have been outlined in the region ... Hydrothermal fluids focused along the major faults, particularly where they are intersected by local northeasterly trending structures."

The Toodoggone volcanics have been the focus of regional study because most of the defined mineral deposits occur in Toodoggone Group rocks. One of several exceptions is the Baker Mine group of deposits. The Baker Mine (Chappelle) mineralization occurs in steep-dipping northeasterly-trending quartz vein systems within a window of Upper Triassic Takla Group volcanics (Barr, 1978). These volcanics are overlain unconformably by Toodoggone volcanics and are cut by Omineca Intrusions varying in composition from granodiorite to quartz monzonite, and locally syenomonzonite. Mineralization, consisting of pyrite, chalcopyrite, electrum, argentite, bornite and sphalerite grains and blebs in brecciated quartz veining, is related to intrusive emplacement and is controlled locally by northeasterly-trending faulting.

Mineralization in the Lawyers and Shasta deposits occur in Toodoggone volcanics as native silver, electrum and sulfides in brecciated quartz veining controlled by steep-dipping northwesterly-trending faulting (Vulimiri, 1983, Downing and Hoffman, 1986).

3.2 Property Geology

The Graves 1 and 2 claims were staked primarily over Early Jurassic Hazelton Group volcanic and volcaniclastic rocks forming the eastern limb of a north-northwest trending faulted anticline (Figure 4). Forster (1984) divided these rocks into four units, which on the property dip steeply to the northeast at 50 to 80 degrees, and form Mt. Graves and adjoining ridges.

Among the Hazelton Group lower units is a welded to partlywelded andesite and pumice lapilli breccia (Unit 7) composed of 30 to 60% clasts set within a vitrified groundmass of lithic fragments, glass shards and crystals of plagioclase, biotite, hornblende and magnetite. Unit 7 was formed as a pyroclastic flow and is unsorted. The clasts range in size from 1/2 to 10 cm, and are rounded, except near the top of the unit where angular, crowded clasts occur. Colour is generally brown to maroon with a less common greenish groundmass near the top of the unit.

Grey, green to orange hornblende porphyritic andesite flows (Unit 8) may be interbedded with the above breccia unit in places. A total thickness for the flows of 425 m was estimated by Forster. Composing the resistant heights of Mt. Graves and the western slopes, this succession of medium-grained flows contain 30-50% plagioclase and hornblende phenocrysts with a similar composition groundmass. Also included in the groundmass are magnetite, hematite and volcanic glass. Some discontinuous andesite tuffs are also present in Unit 8.

Unit 9 is composed of thin, discontinuous, volcanic-derived greywacke with some laminated siltstone. Occurring only on the south shoulder of Mt. Graves at the 1970 m level, this brown to grey sedimentary unit is less than 15 m thick, and displays small scale, recumbent, epigenetic folding.

The uppermost Hazelton Group unit is a series of maroon to brown pyroxene andesite flows (Unit 10) forming resistant caps to ridges north and east of the Mt. Graves summit. Variable thicknesses of crystal and lapilli tuff and agglomerate may mark flow boundaries.

At the core of a northwest-trending anticline on the west margin of the property is an elongate quartz monzonite stock (Unit 6) in fault contact with lower Hazelton Group units. Some related steep-dipping, quartz-monzonite dikes occur on the western half of the property, mostly along northeastly to easterly-trending faults.

Quartz-feldspar porphyry dykes (Unit 22) are steep (55° to near vertical), easterly dipping and trend predominantly northwesterly, subparallel to Hazelton Group bedding. Rhyolite in composition, these dykes consist of potash feldspar, quartz, plagioclase and less than 1% hornblende phenocrysts in a crypto-crystalline quartz-potash feldspar groundmass. Appearing as brown to flesh-coloured "fresh" rock, this unit was found by Forster in thin-section to display pseudomorphs of sericite and carbonate after all components. The resistant, unbroken nature of these dykes leaves them relatively impermeable, and as demonstrated on the GWP Zone, they act to confine and channel mineralizing fluids to broken or more permeable rock.

Pyroxene andesite dykes (Unit 23) are commonly mapped in the region as thin basalt or mafic dykes, and in most cases appear to post-date mineralization. However, on the lower GWP Zone, one thicker (to 2 m) dyke occurs along a mineralized fault and may itself contain mineralization. This is probably a local effect, because dykes of identical composition and orientation cutting upper GWP Zone mineralization were clearly emplaced after the mineralization. These andesite dykes contain up to 50% fine grained plagioclase and clinopyroxene phenocrysts in a plagioclase, chlorite and magnetite groundmass. Small, sparse calcite-filled amygdules are characteristic of late alteration.

Three types of faults provide the dominate structural control and, additionally, control hydrothermal systems on the Mount Graves property. First of these are the highangle northwesterly-oriented block faults which are part of the regional fabric. The best example is the Yellow Rose Fault that persists along the entire southwestern flank of the Mount Graves ridge. Secondly, bedding-plane faults are northwesterly-trending, relatively high-angle (50 to 80), and probably closely related to block faulting. These zones are the preferred intrusive conduit for rhyolite and mafic dykes north and east of the Mount Graves ridgeline. Finally, easterly to northeasterly-trending, high angle faults crosscut northwesterly-trending structures and commonly have a right-lateral component. The mineralized Upper GWP Zone fault is one example. Quartz veining, mineralization, and varying alteration have been found associated with all three of these fault types.

3.3 Alteration and Mineralization

Forster (1984) notes that propylitic alteration in varying intensities is pervasive in all major units, and is seen in thin sections in otherwise fresh appearing rock. Typically, mafic phenocrysts are altered to chlorite and epidote with hematite or magnetite inclusions and rims. Plagioclase phenocrysts are replaced and sometimes zoned with epidote, chlorite and carbonate. Groundmasses are commonly altered to epidote, chlorite, carbonate, sericite and pyrite. Goodall (1984) similarly noted a regional alteration facies of "propylitic to argillic" at the Shasta property. In addition, argillic to advanced phyllic and potassic to silicic alteration facies were found associated with quartz veining and brecciation on the Shasta property. At the Mount Graves prospects these last two alteration facies are either poorly defined or absent. The reasons for this are dealt with later in this section in the discussion of Mount Graves' position in a regional epithermal model.

Highly variable silicification is specifically related to fault conduits and fracture zones, and is found in increasing intensity from hairline veinlets of quartz and calcite through all intermediate stages to complete quartz flooding. Multiple movement along these zones with renewed silicification has resulted in banded and/or brecciated veins in some areas. Late stage veining, consisting of calcite with lesser zeolite and prehnite, is especially common in the propylitized margins of silicified zones.

Pyrite is commonly associated with all intensities of quartz veining as vein-fillings and disseminations especially on partly silicified, clay-altered vein margins. Weathering of pyrite results in red, orange and yellowcoloured zones of iron oxides, jarosite and manganese oxides, which invariably occur along fault and fracture zones. Stockwork and breccia zones are composed primarily of amethystine, clear, white and creamy chalcedonic quartz, with lesser calcite, siderite, hematite and barite. Forster distinguishes five successive stages of mineralization and brecciation:

 Amethystine and white quartz with disseminated pyrite. Veins to 2.5 cm wide.

- 2 & 3. White quartz with minor barite, galena and sphalerite.
 - Banded chalcedonic, amethystine and white quartz with argentite (acanthite), native gold (electrum), native silver, tetrahedrite and pyrite. Veinlets commonly 0.5 cm or less wide.
 - Late stage calcite, siderite, albite and minor barite.

Forster's polished section studies revealed that native gold is associated with at least two stages of mineralization as intergrowths with pyrite, and, in addition, is associated with native silver and argentite.

Forster (1986, p. 135) also conducted a fluid inclusion analysis on mineralized rock from Mount Graves and the Moosehorn Creek property, 16 km west. Based on that study, he proposed a Toodoggone regional epithermal model (Figure This model requires a throttle point at 100 m depth 4a). to provide a best fit to temperature and pressure data. The concept of a throttle consists of a constriction in the mineralization plumbing system that is gradually closed by near-surface deposition of calcite, zeolite, clay and silica gangue minerals. This throttling down permits buildup of pressures greater than a simple hydrostatic column, just as a pressure cooker lid permits pressures greater than atmospheric pressure. A throttled system in turn allows assumption of lithostatic pressures in determining depth of formation. Samples from the GWP Zone of Mount Graves were calculated to have been formed at 225 m depth at the top of the boiling zone with temperatures ranging from 245 to 296°C. Repeated breaching of the throttled system by a tectonic or hyrofracturing mechanism



would allow episodic boiling and thus mineral deposition, at greater depths. Forster concludes:

"Boiling generally occurs near the top of the base metal horizon, in a highly mineralized zone of mixed precious and base metals... The Mount Graves prospect would have probably formed in this mixed zone, as is indicated by the presence of fluid inclusion boiling textures, and the occurrence of galena and sphalerite, as well as gold and silver."

Six separate areas are discussed in following sections of text. The location of all figures accompanying each discussion may be found on the Index Map to property mineral showings (Figure 5). Also shown on the Index Map are the location and points of view of seven photo Figures (5a through 5g) referred to in the text).

3.4 GWP Zone Geology, Alteration and Mineralization

The GWP Zone is composed of two separate but probably related showings, termed the Upper and Lower GWP Zones (Figures 5a and 6).

The Lower GWP Zone lies in a steep-dipping andesite flow forming the hanging wall of a 30 m thick rhyolite (quartz feldspar) dyke (Figures 5b and 5c). Both the mineralized andesite and the rhyolite dykes are cut by a pyroxene andesite dyke 1/2 m to 2 m thick, injected along bedding plane faulting in the andesite. Relict textures of the andesite are preserved except in sparse areas of intense silicification. Fine to medium grained disseminated pyrite composes up to 3% of the rock, especially in areas of increased quartz veining. Some trace galena is also associated with quartz veining. Limonite, jarosite and alunite crusts are present on weathered surfaces. Mineralization





FIGURE 5a. PANORAMA OF THE UPPER AND LOWER GWP ZONES, looking west.



FIGURE 5b. LOWER GWP ZONE, looking northwest.

Note contact of mineralized andesite with rhyolite dyke on left of photo.



FIGURE 5c. GWP ZONE, looking south from hill above Square Lake.





FIGURE 5e. YELLOW ROSE FAULT ZONE, looking along fault to the southeast from the Orange Rose Zone toward the Yellow Rose Zone.



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FIGURE 5f. PANORAMA OF THE YELLOW ROSE FAULT ZONE, looking east. Orange Rose Zone on the left and Yellow Rose Zone on the right.



FIGURE 5g. EAST RIDGE ZONE, looking west from helicopter.



FIGURE 5h. EAST RIDGE, looking southeast from hill above Square Lake.

The East Ridge Zone is near the end of the ridge.

LEGEND

(modified from Forster, 1986)

EARLY TO MIDDLE JURASSIC Hazelton Group

Maroon to brown pyroxene andesite flows, minor crystal 10 and lapilli tuff and agglomerate between flows 10a Green hornblende plagioclase porphyry flows Brown to grey greywacke and siltstone 9 Grey to green hornblende porphyritic andesite flows, 8 minor tan to brown andesite tuffs 8a Maroon to purple porphyritic basalt Grey to green hornblende porphyritic andesite breccia, 8b variable brecciation and silicification INTRUSIVE ROCKS TERTIARY (?) Pyroxene andesite dykes (mafic dykes), calcite-filled 23 amygdules EARLY TO MIDDLE JURASSIC 22 Quartz-feldspar porphyry (rhyolite) dykes Geologic contact, dashed where inferred. Fault, queried where location uncertain Fault with lateral movement Bedding attitude with dip Jointing, vertical, inclined ; Outcrop Altered area Survey point + _2000- Contour, interval 20 m 219,53.0,119,1105 ppb Au ppm Ar Ph Tocation A-659 A Soil sample number and location 540,79.6, 1206, 990 ppb Au, ppm Ag, Pb, Zn


and alteration appear to be controlled by faulting and fracturing subparallel to bedding with little brecciation noted.

The mineralization and alteration do not persist along strike but narrow and disappear just northwest of sample A-661.

The Upper GWP Zone is a steeply northwesterly-dipping fault zone at least 30 m wide striking 030° (Figures 5a, 5d and Movement on the fault from the evidence of striations 6). on the hanging wall is largely right lateral. The fault zone is interbraided with highly variable silicification, brecciation and mineralization. Some small lenses (samples L-342, L-347) are more lightly silicified and retain relict andesite porphyry textures. Most of the samples in Trench 3, however have been repeatedly brecciated and silicified. The more heavily silicified rock may not necessarily contain higher precious and base metal values. Quartz vein banding is not common. Cycles of sugary quartz replacement and fine quartz veining alternate with brecciation, largely destroying relict textures. Pyrite, galena, sphalerite and tetrahedrite are found as blebs, disseminations and stringers in breccia openings. Most breccia fragments show rounding. It is important to note that rounded rhyolite dyke fragments are common. Post-mineral mafic dykes have been emplaced along faults parallel to the andesite bedding planes and cut all earlier rocks including the GWP Zone breccia.

In Trench 4 to the southwest of Trench 3, similar mineralized and altered andesite has been found under 3 m of overburden, indicating a probable extension to the southwest (Figures 6 and 6c). The northeasterly extension of the Upper GWP Zone lies under talus cover probably exceeding 3 m. The only outcrop along the extension is a small mafic



dyke. Trench 5, dug along that dyke, is less than 1.5 m deep and did not cut additional bedrock. Two soil samples (L-659, L-660) taken in Trench 5 were anomalous in gold and silver as may be expected downslope from the Upper GWP showing.

The question of the relationship between the Upper and Lower GWP Zones remains open because of the lack of evidence in the field. However, the presence of altered rhyolite dyke fragments in the Upper Zone suggests that the northeasterly extension does indeed cut through the rhyolite dyke between the Upper and Lower Zones.

3.5 Yellow Rose Fault Zone Geology

The Yellow Rose Fault Zone, with an exposed length of roughly 2 km, is one of the most persistent structures found on the Mount Graves property (Figures 4, 5e and 5f). Air photos most clearly show the northwesterly striking trace of a broadly arcuate, near vertical fault inclined steeply to the southwest. On the extreme northwest ridge of Mount Graves, the fault pinches to less than 10 m wide and is associated with only scattered quartz and calcite veins. However, on the Orange Rose and Yellow Rose gossans, the fault zone is over 30 m wide with associated argillic and lesser quartz-pyrite alteration. One additional gossanous area occurs along the length of the fault in the headwall of the South Bowl, and may be related to the Yellow Rose Gossan. This relation is discussed further in the Geochemistry chapter.

In Trench 1 on the Yellow Rose Zone, sparse, thin and largely barren, clear quartz veining, cuts broken, clayaltered andesite on the explosed hanging wall (southwest). An additional 5 m length of shattered, increasingly clayaltered andesite was excavated further into the fault zone to the northeast during the 1987 program (Figure 7). Thin, sparse quartz veining with little or no sulfides persists in the newly sampled part of Trench 1. With the exception of thin pyritic silicified andesite lenses found in rubble on the northeast side of the fault and a few rusty vein selvages, very little pyrite is available to be oxidized on the Yellow Rose Zone gossan.

In contrast, the Orange Rose Zone contains lenses of moderately to heavily silicified andesite varying from a few centimeters to 2 m thick with up to 5% disseminated pyrite (Figure 7a). Most of the fault is clay altered, ranging from shattered, in-place rock with clay rims to thick gouge. Nowhere along the Yellow Rose Fault Zone was there found the repeated brecciation and silicification associated with the GWP Zone.

3.6 East Ridge and Lake Fault Zone Geology

The East Ridge Zone is a discontinuous series of gossanous areas associated with steep faults cutting moderately easterly-dipping andesite flows, and also with permeable beds between flows. Figure 5g is an oblique aerial photo looking toward the west at the East Ridge Zone. The orange and yellow gossans, such as those in the photo, were sampled by Great Western Petroleum geologists with low, erratic precious and base metal values (Caira, 1982).

Sampling during the 1987 program was confined to rocks along the main ENE-trending fault which pinches and swells along its short exposed length (Figure 8). Four samples were taken in an area of argillic alteration up to 20 m wide. The highest precious metal values were associated with thin quartz-pyrite altered lenses. Calcite veining is common on the periphery of gossanous areas. Several meters downhill an agglomerate up to 4 m thick is exposed that is







partly silicified with some disseminated pyrite and galena. The one sample (L-323) taken had low precious metal values, but contained 3628 ppm Ba. A NNW-trending fault also displayed some quartz-pyrite alteration, but was not sampled. The area merits further prospecting, and permeable horizons such as the agglomerate may offer the best chance of hosting commercial values.

The Lake Fault Zone can be seen in photo Figure 5g in the distance just beyond Square Lake as a series of ENE-trending steps in the pine covered hillside. Rock samples with up to 1230 ppm Au and 37 ppm Ag were collected by Great Western Petroleum geologists just northwest of the lake (Caira, 1982). The sites sampled were probably thin (less than 4 cm) quartz veins with no sulfides which were found along the northernmost pair of steep ENE-trending faults shown in Figure 8. The stair-step faulting found to the south has thin clear to white quartz veining along and cross-cutting the ENE-trending structure, and some galena and sphalerite were found associated. The veining was sampled along one fault and returned disappointing precious metal values of up to 56 ppb Au and 5.7 ppm Ag. The hillside is closely covered with vegetation and has very limited exposure. Boulders found below the hillside have higher concentrations of base metals that were not found in place during the program.

4.0 GEOCHEMICAL SURVEYS

4.1 The 1987 Program

Previous work on the property had outlined three geochemically anomalous zones: the GWP Zone, the Yellow Rose Zone and the East Ridge Zone. During the current field season an effort was made to extend the first two zones, with soil and stream sampling on small grids and with reconnaissance soil and stream sampling along strike and in the general vicinity. Also, additional mapping, rock chip sampling and hand trenching were concentrated in these areas. A more limited amount of mapping and rock sampling was done on the East Ridge Zone.

Two other areas received brief attention: the South Wall Gossan and the Lake Fault Zone. Because of an obvious gossan and previous reconnaissance soil sampling of anomalies below the South Wall area, a small soil sampling grid was established on the ridge top above the gossan. A new mineralized area, the Lake Fault Zone was found by prospecting, and limited rock chip sampling was conducted.

Each of the above zones are detailed in subsequent sections complete with relevant geochemical results.

4.2 Sampling and Analytical Procedures

A total of 116 soil samples were dug with mattocks or soil augers, and collected in Kraft paper bags. Except for rocky terrain, a minimum sampling depth of 20 cm was maintained, collecting "B" horizon soil. A total of 10 stream samples were taken from active stream sediments and dry stream beds, both along grid lines and while prospecting. The 57 rock samples collected are described in Appendix IV and are graded as channel, rock chip and grab samples according to decreasing reliability of representation. These classifications are indicated as a suffix to rock sample numbers as "C", "R" and "G" respectively.

All samples were sent to Min-En Laboratories Ltd., 705 West 15th Street, North Vancouver, B.C. for analysis. All samples were subjected to a 12 element ICP analysis for Ag, As, Ba, Cd, Co, Cr, Cu, Mn, Mo, Pb, Sb and Zn. All samples were analysed for Au using atomic absorption methods for soil and stream samples, and geochemical methods with fire assay preconcentration and atomic absorption finish for rock samples.

Soil and stream samples were dried at 95°C, then sieved to separate the minus 80 mesh fraction. A 1 gm portion of this fraction was placed in a test tube and digested for 6 hours with 1:1 equimolar (50%) aqua regia. After cooling, samples were diluted to a standard volume, and the solution analysed using a Jarell Ash model 900ICP Inductively-Coupled Plasma Analyser.

Rock samples were crushed and split, separating a 300 g pulp. For ICP analysis the pulp is pulverized by ceramic plate pulverizer to minus 80 mesh and processed the same as soils. For geochemical analysis for gold, a 300 g split is pulverized to minus 150 mesh, and a 15 g sample weight is fire assay preconcentrated. The sample is then digested with aqua regia and taken up with 25% HCl. The gold is extracted with methyl iso-butyl ketone, and analysed by atomic absorption to a detection limit of 1 ppb against a standard gold solution.

4.3 Discussion of Geochemical Results

Reports tabulating analytical results are presented in Appendix V.

Because of the non-random location of samples on the various mineral occurrences and the limited number of samples taken in each area, conventional statistical analyses with assignment of anomalous values is not appropriate. Some correlation coefficients were calculated for selected metal relations in separate zones, and are discussed below. Rock chip sampling on the GWP Zone yielded results for representative sampling of up to 3000 ppb Au, 503 ppm Ag, 14,493 ppm Pb and 33,791 ppm Zn. In Trench 3, eleven rock chip samples (L-341 to L-351) averaged 971 ppb Au, 272 ppm Ag, 3312 ppm Pb and 4880 ppm Zn on a triangular exposure measuring 5 m wide by 8 m along strike. These results are consistent with earlier reported sampling in the same trench of up to 0.170 oz/ton Au and 149.21 oz/ton Ag (Yeager & Ikona, 1986) and up to 11,000 ppb Au, 7500 ppm 30,000 ppm Pb and 33,000 ppm Zn (Caira, Aq, 1982). Analysis of trench soil profiles proved that soil sampling to detect mineralized extensions in the GWP Bowl would not be effective unless samples were collected from at least 180 cm (6 feet) depth in soil flow and talus overburden exceeding 3 meters.

4.4 GWP Zone Geochemistry

On the GWP Zone, 28 rock chip or rock channel samples (L-330 to L-351, A-661 to A-666) were collected with values returning up to 3,000 ppb Au, 503 ppm Ag, 106 ppm As, 373.1 ppm Cd, 1,445 ppm Cu, 5,503 ppm Mn, 14,493 ppm Pb and 33,791 ppm Zn (Figure 6c). Correlation coefficients (r) calculated for selected metal pairs using analyses from the 28 GWP zone rock samples are tabulated below:

	Ag	Au	Cđ	Cu	Pb	Zn
Ag	1.00	0.88	0.45	0.59	0.62	0.45
Au		1.00	0.73	0.76	0.83	0.74
Cd			1.00	0.94	0.93	0.99
Cu				1.00	0.95	0.93
Pb					1.00	0.94
Zn						1.00

The strong cadmium-zinc correlation indicates a consistent substitution of cadmium in sphalerite. The high lead-zinc correlation implies that at least locally, mineralization was coeval. The more marginal silver-lead and silvercopper coefficients in conjunction with the high copperlead coefficient suggest that silver mineralization is not exclusively tied to either galena or tetrahedrite deposition, and that argentite and/or native silver may be important contributors to silver values. Additional mineralogical evidence may be required to account for the high correlation of copper to cadmium, lead and zinc.

A small soil sample grid was established to test for a southwesterly extension of the GWP Zone mineralization (Figures 6a and 6b). Kowall (1984) had tested the extension with inconclusive results using several cross cutting traverse lines and relatively shallow soil sampling because of adverse weather. In order to improve the method, samples from a minimum depth of 60 cm were collected using soil augers. This technique failed to yield samples with values that could be considered anomalous. The reason becomes apparent when examining soil profile results from Trench 4 (inset Figures 6a and 6b). Within 8 meters southwest from old trenching (Trench 3) on the GWP zone where overburden is less than 1 meter deep, the overburden depth increased to over 3 meters in Trench 4. The soil profile in Trench 4 shows that a doubling of silver and lead values over previous samples does not occur until 180 cm depth (sample A-624). Clearly anomalous metal values are encountered in the next soil sample (A-625) at 240 cm depth. Below 240 cm the soil becomes rusty and sandy between boulders and cobbles of pyritic silicified andesite (rock sample L-339). The same type of pyritic silicified vein bedrock (sample L-440) was finally uncovered at just over 3 meters depth. Values of these Trench 4 rock samples are lower (up to 135 ppm Au, 52.7 ppm Ag, 377 ppm Pb and 504 ppm Zn) than rock chip samples just to the north in Trench 3 (samples L-341 to L-351) which averaged 971 ppb Au, 272 ppm Ag, 3,312 ppm Pb and 4,880 ppm Zn. Nevertheless, the





mineralization, alteration and rock type are identical in Trenches 3 and 4, firmly indicating that the zone extends to the southwest across a late, thin pyroxene andesite dyke and a possible fault encountered at the south end of Trench 3.

Further south in the GWP Bowl, Trenches 6 and 7 (Figure 6) were dug along an easterly-trending fault containing some associated quartz veining and disseminated pyrite. Four rock chip samples (L-315 to L-318) in Trench 7 returned values up to 90 ppb Au and 13.4 ppm Ag.

One further comment on overburden in the GWP Bowl is needed. The colluvial wedge formed at the slope break between the northwestern cirque wall and the bottom of the GWP Bowl is only 3 m deep at Trench 4. At Trench 4, however, there is little upslope area to contribute material to overburden (Figure 5a). As one proceeds further into the Bowl along the southwesterly extension of the GWP Zone, the depth of overburden will increase, and thicknesses of 5 m are probable.

4.5 Yellow Rose Zone Geochemistry

Trenching by Geostar Mining Corp. (Yeager and Ikona, 1986) on the Yellow Rose Zone had encountered gold mineralization in thin quartz veining (sample #36287, 0.306 oz/ton Au in 5 cm vein). This value was obtained at the edge of exposed bedrock, but still within the clay altered fault zone (Figure 7). The trench (now named Trench 1) was deepened and extended, but only to 25 meters length because of overburden slumping and time limitations. Six rock channel samples (L-303 to L-308) were taken from the Trench 1 extension in and adjacent to thin quartz veining, returning values up to 960 ppb Au, 0.6 ppm Ag, 188.7 ppm Cd, 3,554 ppm Pb and 4,932 ppm Zn. Silver values are lower than those reported from the southwest end of the trench by Yeager, which ranged from 0.01 to 0.19 oz/ton Ag. Some mineral zoning may be indicated. A suspected correlation of cadmium and zinc was confirmed by a coefficient (r) of 0.80 calculated on values from the eight samples taken on the zone. Two rock grab samples (L-301 and L-302) taken from rubble on the northeast margin of the fault zone returned values up to 238 ppb Au, 0.9 ppm Ag, 55.9 ppm Cd, 1,721 ppm Pb and 4,561 ppm Zn. These grey, pyritic quartzflooded rocks come from lenses at least 15 cm thick. Apparently at least some intense silicification is associated with argillic alteration on the northeastern side of the Yellow Rose Zone.

Roughly 550 m to the northwest of the Yellow Rose Zone and lying on the same 35 m-wide near vertical fault zone is an orange pyritic gossan (Figure 7a). This second gossan, named the Orange Rose Zone, had not been sampled before, and returned (samples L-309 to L-314) up to 23 ppb Au, 2.9 ppm Ag, 185 ppm As, 159 ppm Pb and 126 ppm Zn, in rock chip and channel samples from outcrop and trenching (Trench 2). The samples were taken from thin, moderately to highly silicified andesite lenses with up to 5% disseminated pyrite.

A southeasterly extension of the Yellow Rose Zone (Figures 7b and 7c) was tested with three soil sample crosslines spaced 50 m apart, measuring from Trench 1 along the Yellow Rose Fault. Deep soil samples (samples A-697 to A-708), collected at 60 cm depth, from tan to white clays similar to those in Trench 1, returned values ranging from 5 to 20 ppb Au, 0.3 to 5.9 ppm Ag, 187 to 1287 ppm Ba, 2.2 to 13.8 ppm Cd, 44 to 418 ppm Pb and 27 to 662 ppm Zn. Similar to Trench 1 rock samples, a positive cadmium-zinc correlation coefficient of 0.91 was calculated for the 12 soil samples on the Yellow Rose Zone. Because of this high correlation







and anomalous lead and zinc values, mineralization in Trench 1 and on the southeast extension are probably related. The anomalous silver values may be related to mineralization from an intersected system in the South Bowl or from a part of the Yellow Rose Zone not intersected by Trench 1.

The fault zone from Trench 1 on the Yellow Rose Zone to Trench 2 on the Orange Rose Zone (Figures 7b and 7c) was tested with reconnaissance stream and soil sampling (samples K-101 to K-122). The method was to take a sample in the fault zone near the downhill side, which is marked by a slope break and greener vegetation. A second sample was taken 40 m uphill above the fault zone, and analyses of the two samples compared. No clear-cut contrasts are found in the eleven sample pairs collected. The best contrast is between samples K-117 and K-118:

Above Fault

On Fault

K-118	5.0	ppb	Au	K-117	35.0	ppb	Au
	0.9	ppm	Ag		1.9	ppm	Ag
3	75.0	ppm	Pb		228.0	ppm	Pb
	169.0	ppm	Zn		400.0	ppm	Zn

Some anomalous values associated with gossans in the Vole Peak vicinity may be expected on the southwestern slopes.

4.6 East Ridge and Lake Fault Zone Geochemistry

The East Ridge Zone in the northeast corner of the Graves 1 claim was sampled by Great Western Petroleum Corp. geologists on traverse along the 1650 meter contour (Caira 1982). Soil sample values up to 50 ppb Au, 8.2 ppm Ag, 1600 ppm Pb and 495 ppm Zn, and rock chip sample values up to 30 ppb Au, 2.7 ppm Ag, 335 ppm Pb and 211 ppm Zn were obtained in the vicinity of gossans and associated pyritic silicified andesite. During the 1987 program four rock samples were taken from pyritic silicified vein rock and andesite vein margins where an ENE-trending vertical fault zone up to 20 m wide crosses the ridge crest (Figure 8). One sample (L-319) of 25 cm thick grey quartz-flooded vein rock with 3 to 5% disseminated pyrite returned values of 290 ppb Au, 55.6 ppm Ag, 419 ppm Pb and 412 ppm Zn. One additional sample (L-323) taken in pyritic, partly silicified agglomerate at a flow boundary yielded up to 23 ppb Au, 0.3 ppm Ag, 508 ppm Pb, 107 ppm Zn and 3628 ppm Ba.

On the Lake Fault Zone, samples of quartz veining along one of the faults cutting andesite porphyry flows returned values up to 56 ppb Au, 6.2 ppm Ag, 218 ppm Pb and 267 ppm Zn. The source of float containing higher base metal concentrations was not found.

4.7 East Slope Vole Peak Geochemistry

Traverses along the 1850 meter and 1650 meter contours by Great Western Petroleum geologists in 1981, returned several anomalous soil samples on the east slope of Vole Peak in the South Bowl (Caira 1982). Soil values up to 85 ppb Au, 8.4 ppm Ag, 345 ppm Pb and 615 ppm Zn were reported. During the current program (Figures 9 and 9a) two short follow-up traverses collecting soil on 15 m spacing were made on the 1800 meter and 1750 meter con-Samples were taken from 60 cm depth or greater tours. using soil augers, except in very rocky ground. Analysis of the soils (samples A-627 to A-654) returned values up to 35 ppm Au, 7.1 ppm Ag, 653 ppm Pb and 372 ppm Zn. A small area of lead values greater than 400 ppm can be outlined on the 1750 contour traverse. No related mineralization was found in the limited outcrop in the vicinity.







4.8 South Wall Gossan Geochemistry

In 1981, Great Western Petroleum geologists noticed the South Wall gossan near the rim of the southernmost cirque (South Bowl) on the Graves 1 claim. Rock sampling along the ridge above the gossan produced near background values of up to 5 ppb Au, 1.8 ppm Ag, 70 ppm Cu, 19 ppm PB and 174 ppm Zn (Caira 1982). However, soil sampling on the 1650 contour in the South Bowl below the gossan produced values of up to 80 ppb Au, 3.7 ppm Ag, 265 ppm Pb and 522 ppm Zn.

During the 1987 program, one man-day was spent collecting soil samples on three short lines to test the ground on the cirque rim directly over the gossan and to test a possible connection to another gossan 350 m southeast along the east border of the Graves 1 claim (Figures 10 and 10a). Directly over the gossan (samples A-669 to A-675) values were returned up to 5 ppm Au, 5.8 ppm Ag, 302 ppm Pb and The remaining lines, run roughly along the 254 ppm Zn. 1990 and 1970 meter contours (samples A-678 to A-696) on the south slope of the ridge, produced values up to 15 ppb Au, 1.1 ppm Ag, 4.8 ppm Cd, 104 ppm Pb and 254 ppm Zn. The highest precious and base metal results lie down slope to the south of the South Wall gossan, and no connection with the other gossanous area to the southeast is apparent from the above results.

5.0 DISCUSSION AND CONCLUSIONS

The Mt. Graves property has been the subject of gold exploration since 1980 because of its proximity to the Lawyer, Baker and Shasta properties, and because of similarities in mineralization and alteration to those properties. Forster (1984), in an M.Sc. thesis, detailed these similarities and provided fluid inclusion evidence placing the property within a favourable environment for precious







and base metal deposition based on epithermal gold models. Control of alteration and mineralization is exercised primarily by profound high-angle faulting, but on the property, some channeling of mineralizing fluids has also been accomplished by extensive quartz feldspar porphyry dykes and less permeable andesite flows. Easterly to northeasterly high-angle faulting also hosts mineralization, commonly at or near intersection with northwesterly-trending faults.

Geochemical anomalies reported by Kowall (1984), Caira (1982) and Yeager & Ikona (1986) were generally confirmed during the 1987 program by trenching, and rock and soil sampling results. Mapping and trenching on the GWP zone has indicated a favourably mineralized southwestern extension along strike that should be tested by drilling. Mapping also indicates a probable northeast extension. Sampling and mapping on the Yellow Rose Zone indicates that northerly and southerly extensions may exist along a steep northwesterly-trending shear zone. Mapping and sampling on the East Ridge Zone has demonstrated that steep faulting and an overlying, relatively impermeable andesite flow control mineralization. The Lake Fault Zone is associated with guartz-galena-sphalerite veining, and deserves additional prospecting. Several other anomalous zones and gossans merit further prospecting.

6.0 RECOMMENDATIONS

The following program is recommended for the next stage of exploration:

 Drilling should be conducted on southwesterly and northeasterly extensions of the mineralized quartz breccia zone of the GWP zone. This drilling should be completed during the period of mid-June to mid-July to take advantage of limited water from snow melt in the upper GWP cirque.

- 2. Simultaneous to drilling, geophysical surveying, minimally a ground VLF-EM and magnetic survey, and further mapping and trenching should be conducted on the GWP zone to define extensions. Detailed land survey techniques should be used to locate all work past and present in this area and to establish elevation controls.
- Additional trenching, mapping, and rock and soil sampling are required on the Yellow Rose and East Ridge Zones to define mineralized areas and structural controls.
- Further mapping, prospecting and sampling are required to evaluate the Lake Fault and the South Wall areas.

A cost estimate for the above proposed program is included in Appendix I.

Respectfully submitted,

HI-TEC RESOURCE MANAGEMENT LTD.

David A. JLyman, Geological Engineer

February 29, 1988

7.0 REFERENCES

- Barr, D.A. 1978. The Chappelle Gold-Silver Deposit, British Columbia; CIM Bull., Vol. 71, pp. 66-79.
- Caira, N.M. 1982. Geological and Geochemical Report Graves 1-8 Claims; B.C. Ministry of Energy, Mines and Petroleum Resources (BCMEMPR) Assessment Report #10050.
- Carter, N.C. 1972. Toodoggone River Area; B.C. Ministry of Energy, Mines and Petroleum Res. (BCMEMPR), G.E.M. 1971, pp. 63-70.
- Diakow, L.J. 1984. Geology between Toodoggone and Chuckachida Rivers (94E), in Geological Fieldwork 1983; Paper 1984-1, pp. 139-145.
- Diakow, L.J., Panteleyev, A., and Schroeter, T.G. 1985. Geology of the Toodoggone River Area; BCMEMPR, Preliminary Map 61.
- Downing, B.W. and Hoffman, S.J. 1986. A Multidisciplinary Case History of the Shasta Epithermal Gold-Silver Deposit, B.C., Canada, in Geoexpo/86, Exploration in the North American Cordillera, eds. Elliot, I.L. and Smee, B.W.; Assoc. of Explor. Geochemists, Vancouver Symposium, May 1986.
- Forster, D.B. 1984. Geology, Petrology and Precious Metal Mineralization, Toodoggone River Area; University of British Columbia, unpublished M.Sc., Thesis.
- Goodall, J.A. 1984. Geology and Alteration of the Toodoggone Volcanics, Shasta Property; University of British Columbia, unpublished B.Sc. Thesis.
- Kowall, C. 1984. A Geochemical Sampling Report Covering the Mt. Graves Mineral Claims, Omineca Mining Division, B.C.; BCMEMPR Assessment Report #13458.
- Panteleyev, A. 1983. Geology between the Toodoggone and Sturdee Rivers (94E), <u>in</u> Geological Fieldwork, 1982; BCMEMPR, Paper 1983-1, pp. 142-148.
- Panteleyev, A. 1984. Stratigraphic Position in the Toodoggone Volcanics, <u>in</u> Geological Fieldwork 1983; BCMEMPR Paper 1984-1, pp. 136-138.

Schroeter, T.G., Toodoggone River Area (94E); BCMEMPR, 1981, Geological Fieldwork 1980, Paper 1981-1, pp. 124-131 1982, Geological Fieldwork 1981, Paper 1982-1, pp. 122-123 1983, Geological Fieldwork 1982, Paper 1983-1, pp. 125-133 1984, Geological Fieldwork 1983, Paper 1984-1, pp. 134-135 1985, Geological Fieldwork 1984, Paper 1985-1, pp. 291-297

Schroeter, T.G. and Lefebure, D.V. 1987. Toodoggone River Area (94E), Geological Fieldwork 1986, BCMEMPR, Paper 1987-1, pp. 111-121.

- Schroeter, T.G., Panteleyev, A. and Diakow, L.J. 1986. Epithermal precious metal deposits in Toodoggone River area, B.C. - implications for deposits in Mesozoic and older volcanic terranes; paper presented at Gold 86, International Symposium on Geology of Gold Deposits, Toronto, September 1986.
- Vulimiri, M.R., Tegard, P. and Stammers, M.A. 1983. Lawyers Gold-Silver Deposit, British Columbia; paper presented CIM District 6 Meeting, Smithers, B.C., Oct. 26-29, 1983.
- Yeager, D.A. and Ikona, C.K. 1986. Geological and Geochemical Report on the Mt. Graves Property; BCMEMPR Assessment Report #14824.



APPENDIX I

Estimated Cost of Proposed Program

Project Preparation	\$	1,500.00	
Mobilization/Demobilization			10,000.00
Personnel			
1 Project Geologist	CE 500 00		
2 Geological Technicians	\$5,500.00		
20 days @ \$225.00/day	9,000.00		
1 Geophysical Technician			
3 days @ \$250.00/day	750.00		15 000 00
			15,000.00
Diamond Drilling (all inclusive)			
500 m @ \$150.00/m			45,000.00
Helicopter Transport - 20 hours @		11,000.00	
Trenching and Survey equipment rea		3,000.00	
Geophysical Equipment Rental			2,000.00
Geochemical Analyses - 350 samples	s @ \$11.20		3,920.00
Camp Equipment Rental - 20 days @	\$80.00/day		1,600.00
Domicile - 63 man days @ \$30.00/da	ау		1,890.00
Field Supplies		3,500.00	
Report Costs			4,500.00
Project Supervision and Engineering	-	5,500.00	
		\$1	08,660.00
10% Contingency		-	10,866.00
	TOTAL:	\$1	19,526.00



APPENDIX II

STATEMENT OF COSTS

BLUE EMERALD RESOURCES INC. - MT. GRAVES PROPERTY Project 87BC030 September 2 to 18, 1987

\$ 1,500.00 Project Preparation Salaries D. Lyman, Project Geologist 17 days @ \$375.00/day \$6,375.00 A. Cooper, Field Technician 17 days @ \$200.00/day 3,400.00 K. Curry, Field Technician 3,400.00 17 days @ \$200.00/day M. Carson, Field Technician 2,600.00 13 days @ \$200.00/day 15,775.00 Supervision J.P. Sorbara, Geologist 2,000.00 4 days @ \$500.00/day Domicile - 68 man days @ \$50.00/day 3,400.00 12,029.66 Mobilization/Demobilization 1,174.75 Support Flights, Helicopter 2 hrs. 4,250.00 Camp Rental - 17 days @ \$250.00/day Field Equipment Rental - 17 days @ \$100.00/day 1,700.00 Geochemical Analysis - 12 element ICP, F.A. Au @ \$10.90/sample \$1,264.40 116 soil samples 10 stream samples @ \$10.90/sample 109.00 57 rock samples @ \$15.25/sample 869.25 2,242.64 Radio Rental - 17 days @ \$30.00/day 510.00 4,500.00 Assessment Report 49,082.06 19% Project Management Fee 9,217.94 J.P. Sorbara & Associates Engineering Report 2,500.00

TOTAL: \$60,800.00

.

APPENDIX III

Statement of Qualifications



STATEMENT OF QUALIFICATIONS

- I, DAVID A. LYMAN, of Vancouver, British Columbia, certify that:
- I am employed as a geologist by Hi-Tec Resource Management Ltd., 1500 - 609 Granville Street, Vancouver, British Columbia.
- I graduated in 1969 from The Colorado School of Mines with the degree of Geological Engineer.
- 3. I have 18 years of experience as a geologist in mineral exploration in Alaska, Canada, the Western United States and Mexico.
- I have neither received nor expect to receive any financial interest, direct or indirect, in the property examined in this report or any property within a 10 km radius.
- 5. This report is based on examinations I personally conducted and work I supervised during August and September 1987, and on geological reports and maps from government, company, and consultant sources, and other professional literature.

eological Engineer David

February 29, 1988

APPENDIX IV

Rock Sample Descriptions


ROCK SAMPLE DESCRIPTIONS

<u>Sample #</u>	Sample Type*	Width (cm)	Rock Description
L-301	G	float	Grey pyrite quartz-flooded porphyry (?) lens with thickness of at least 15 cm indicated.
L-302	G	float	As L-301, tight silicified rock with no oxidation.
L-303	с	30	Footwall selvage of Yeager sample #36287 (see Figure 7), broken, partly-altered andesite.
L-304	с	50	Hanging wall selvage, as L-303, broken, manganese oxide staining.
L-305	с	50	Footwall selvage of thin quartz vein, broken, partly clay-altered andesite, trace chalcopyrite.
L-306	с	45	Hanging wall selvage, as L-305.
L-307	с	1-2	Open quartz vein, no sulfides.
L-308	с	40	Shattered black rusty andesite (?)
L-309	с	40	Highly silicified porphyry (?), 1- 3% very fine grained euredral dis- seminated pyrite, distinct pink-red oxide on surface.
L-310	с	20	Highly silicified porphyry in broken lens, 1-3% pyrite, rusty fractures.
L-311	с	40	As L-310, tightly silicified.
L-312	R	100	Greenish partly-silicified andesite porphyry, approx. 1% euhedral dis- seminated pyrite, very fine quartz- calcite veinlets, rubble.
L-313	R	150	Yellow to red-brown oxidized andesite porphyry (?), very siliceous with 1-5% euhedral dis- seminated pyrite, calcite fracture filling and veinlets.



Sample #	Sample Type*	Width (cm)	Rock Description
L-314	R	200	Highly silicified porphyry with 1- 3% disseminated pyrite, light to heavy oxidation depending on frac- ture density.
L-315	R	170	Moderate-highly silicified porphyry andesite flow (?), >1% disseminated pyrite.
L-316	R	120	Similar L-315, less silicified.
L-317	R	100	Similar L-315, less silicified.
L-318	R	100	Similar L-315, less silicified, some patches of propylitic alter- ation.
L-319	G	25	Grey quartz-flooded lens with 3-5% disseminated pyrite.
L-320	G	40	Rusty weathered clay-quartz frag- ment gouge from shear.
L-321	R	60	Moderate-heavily silicified por- phyry along mineralized fault zone, 1-2% disseminated pyrite, partly oxidized.
L-322	R	80	As L-321, less silicified and pyri- tized fault margin.
L-323	G	40	Partly silicified agglomerate, 3-15 cm diameter rounded cobbles, cut by quartz veinlets.
L-324	R	50	Broken, partly quartz-healed andesite, partly oxidized.
L-325	R	50	As L-324, more silicified, trace galena and sphalerite.
L-326	R	2-4	Quartz vein cross-cutting fault zone, in green andesite.
L-327	R	150	Green porphyry andesite, broken with partial quartz vein healing.
L-328	R	30	Broken, partly quartz-healed andesite along fault trace. Minor pyrite and hematite.

<u>Sample #</u>	Sample Type*	Width (cm)	Rock Description
L-329	R	45	As L-328.
L-330	R	110	Heavily silicified andesite brec- cia, some quartz-feldspar porphyry fragments, galena disseminated and in breccia openings, lesser spha- lerite and pyrite.
L-331	R	150	Similar L-330, repeatedly brec- ciated, less sphalerite and galena.
L-332	R	150	Similar L-331, partly oxidized.
L-333	R	150	Similar L-331, partly oxidized, some less silicified areas.
L-334	R	150	Moderate silicified, partly oxi- dized, some galena along breccia fractures.
L-335	R	150	Heavily silicified andesite and quartz-feldspar porphyry breccia, repeatedly shattered, partly healed, some oxidation 1-3% very fine grained pyrite.
L-336	R	170	Light-moderate silicified andesite breccia, 102% fine grained dissemi- nated pyrite, quartz-feldspar por- phyry intrusive breccia fragments.
L-337	R	200	Moderate-heavily silicified andesite porphyry breccia, with some intensely silicified frag- ments.
L-338	R	250	Similar L-337.
L-339	G	float	Silicified, pyritic andesite from 285 cm depth in Trench 4.
L-340	R	30	Heavily silicified (?), 1-2% very fine grained disseminated pyrite, very fine grained argentite? and galena in clear quartz veinlets, 1 cm oxidized rind, bedrock Trench 4.

<u>Sample #</u>	Sample Type*	Width (cm)	Rock Description
L-341	R	70	Moderate-heavily silicified ande- site, some breccia fragments, 1-3% very fine grained disseminated pyrite after mafics, also fine grained disseminated galena and sphalerite.
L-342	R	50	Light to moderate silicified andesite, <1% disseminated pyrite.
L-343	R	90	Heavily silicified vein rock, repeatedly brecciated, 1-2% very fine grained disseminated pyrite, late stage quartz vein.
L-344	с	5-8	Rusty clay and fragment gouge from small fault, northeast-trending, sample from 25 cm length.
L-345	R	130	Similar to L-343.
L-346	R	130	Similar to L-343.
L-347	R	120	Lightly silicified porphyritic andesite, manganese staining, some fine quartz veining and dissemi- nated pyrite.
L-348	R	120	Similar to L-343.
L-349	R	150	Intensely silicified breccia, relict texture gone, 2-5% very fine grained disseminated pyrite.
L-350	R	150	Simialr to L-343, some amethystine quartz, 3-5% very fine grained euhedral pyrite.
L-351	R	100	Similar to L-349, but more porous and leached.
A-661	R	100	Greenish andesite, moderate silici- fication, thin quartz veinlets trending north-south, some dissemi- nated pyrite.
A-662	R	100	As A-661, increased silicification and pyrite.



Sample #	Sample <u>Type*</u>	Width (cm)	Rock Description
A-663	R	100	Moderately silicified andesite with 1-2% pyrite, brecciated along sub- faults.
A-664	R	100	Thin tight mafic dyke, little altered.
A-665	R	100	Moderate to heavily silicified andesite (?), up to 5% pyrite as dissemination and fracture filling, partly oxidized along fractures, jarosite (?) and alunite crusts.
A-666	R	100	Quartz-feldspar porphyry intrusive, 1 m from contact, unmineralized.
*Sample	Type - C R G	= Chan = Rock = Grab	nel Sample Chip Sample Sample



APPENDIX V

Geochemical Analytical Results



FROJ	ECT NO:	87 BC	030		705	WEST 15TH	ST., NO	RTH VANC	COUVER, B.	C. V78	112		F	ILE NO:	7-1413R	/F1+2
ATTE	WTIDN:	D.A.LYN	AN			(604	1980-531	OR (60	4) 988-452	4	+ TYPE	ROCK BEI	OCHEN +	DATE: 5	EPT 30,	1987
(VA	LUES IN	PPN }	AG	A5	BA	CD	03	CU	NH	MD	PB	58	ZN	CR	AU-PPB	
13	* 10	G	.8	1	436	55.9	11	71	2022	2	695	4	4651	35	29	TRENM
L 3	02	q	. 19	1	201	25.6	16	73	3601	1	1721	3	1124	30	238	incric
L 3	13	ç	. 6	1	592	54.4	4	42	1080	1	213	3	3085	27	16	1/
L 3	94	C	.2	1	134	188.7	5	424	2077	1	136	2	4932	18	9	YE40
L 30	5	C		1	197	28.8	5	238	1458	1	3554	5	1364	63	11	Ros
L 31	6	С	.3	1	123	36.0	6	26	2101	1	129	2	1864	25	6	7
L 3	07	C	. 6	3	78	. 7.0	2	29	471	1	859	2	400	260	940	FONT
L 30	8	C	.3	1	109	16.1	10	23	1943	1	500	2	689	44	4	
L 3	9	C	.7	8	166	3.7	2	7	129	4	83	2	92	97	12	TRENC
L 3	0	<u>c</u>	B	10	233	2.8	2	5	85	4	93	1	75	71	6	2
r 2	1	C	2.9	39	254	2.5	1	5	74	13	159	2	36	121	5	ORAN
L 3	2	R	.8	20	51	3.7	2	5	1254	1	40	2	126	41	6	DAG
L 3	3	R	1.0	30	128	2.6	2	6	456	2	42	1	49	111	11	1405
L 3	4	R	.7	185	242	6.3	2	9	76	4	45	1	26	82	23	ZONE
L 31	5	R	7.6	24	625	4.3	3	16	1278	1	105	2	191	124	33	(In
L 31	6	R	13.4	26	1431	2.4	1	15	109	1	360	3	78	144	90	GWP
L 31	7	R	6.4	37	510	3.0	2	9	422	1	119	3	112	135	54	AREA
L 31	8	R	7.0	41	204	3.9	3	В	819	4	119	1	186	80	62	TRENCH
L 3	9	G	55.6	69	292	8.8	5	40	6852	11	419	3	412	120	290	
L 32	0	G	.7	1	111	6.0	3	6	402	1	23	2	246	75	4	EAST
L 32	1	R	.2	1	113	3.8	2	6	477	1	21	3	175	106	6	D.N.
L 32	2	R	.2	1	58	3.B	2	5	1244	1	24	2	123	59	6	RIUgo
L 37	3	Ġ	.3	3	3628	3.5	5	37	1305	7	50B	1	107	158	23	ZONE
L 32	4	R	4.0	47	406	2.5	4	26	1382	1	86	4	81	146	21	
L 37	5	Ř	5.7	78	193	5.1	4	21	2311	9	179	1	267	217	56	
3	6	R	5.4	66	77	3.7	3	20	3446	5	149	1	117	127	79	TAKE
3	7	P	6.7	101	60	5.2	3	30	8036	6	218	4	145	101	38	CAN
35	B	6	2.7	49	70	3.0	i	17	3481	ĩ	48	7	91	50	7	MOL
1 3	Q.	R	T.8	22	150	3.2	5	18	2451	i	145	ĩ	158	RO	18	ZONE
1 11	0	P	17.6	43	197	37.0	1	135	1475	2	107	-	1175	20	129	
T	1	-D	6.9	76	471	5 5		54	1004		111		107	170	170	
1 7	2	2	3.7	55	7/1	2.7	-	20	1074	4	115	-	175	130	17	
1 1	1	2	7.2	50	107	11.1	2	20	7104	5	104	3	615	274	12	ILDOC
1 7		K	0.2	11	10/		3	57	2179		155		613	170	44	UYPER
1 77	-	K	7.4	02	171	7.1	2	22	3708	4	134	1	100	120	90	GWP
L 33		K	2.6		215	4.3		17	1308	<u></u>	154	3	125	1/4		FONE
L 33	0	R	1.8	12	114	1./	2	28	1265	1	00	3	81	155		P
L 3.		R	1/4.2	36	436	1.2	1	144	2498	1	954	3	491	149	4/0	
L 3.	8	K	11.3	24	128	4.8		20	2418	1	161	1	14/	148	3/	10.11
L 3.	17	9	52.1	54	268	9.0	1	31	2202	1	5//	6	504	92	135	TKENGI
- 34	0	<u>K</u>	18.9	63	152	5.8		14	5106	!	15/	5	228	114	52	4
L 34	1	R	451.5	50	235	373.1	6	2653	3663	6	14493	28	33/91	74	3000	
1 34	2	R	9.0	3	146	9,9	7	125	537	3	299	1	745	97		LUADER
1 34	3	R	234.7	105	429	26.3	1	201	1505	14	1535	13	2171	185	410	GUID
r 34	4	ç	420.6	101	353	77.5	4	1445	5012	5	5519	14	5845	123	800	GWP
1 34	5	<u>R</u>	43.7	60	427	16.2	4	68	1045	7	549	6	717	157	96	TONE
. 34	6	R	228.5	105	325	15.5	3	341	1158	5	1846	9	1067	125	430	maria
L 34	7	R	301.2	72	607	18.7	2	319	1825	4	1781	9	1536	143	990	TKENCI
L 34	8	R	195.6	98	407	27.5	4	277	779	6	1299	_ B	2212	122	940	3
r 31	9	R	140.2	52	592	14.8	2	107	635	7	1413	7	1176	144	570	~
L 35	0	<u>R</u>	451.8	41	270	19.4	2	341	243	3	3916	11	2045		1450	
L 35	1	R	503.1	42	202	25.1	4	392	778	4	3789	9	2380	117	2100	a statute
A 68	1	R	27.0	85	98	9.2	4	67	2251	1	332	4	332	94	34	1
A 68	2	R	53.0	23	1223	18.2	3	251	3586	3	779	5	1105	108	215	LOWE
A 68	3	R	16.1	51	139	4.2	1	19	528	1	142	4	243	47	59	Guit
A 66	4	R	5.2	56	471	3.6	- 1	22	253	1	87	5	69	96	3	YWP
A 66	5	R	562.8	58	186	21.4	6	187	1844	6	697	3	1091	32	1400	ZONE
	1	D	10 0		154		7	5	1493	1	74	1	65	37	1	

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★ SAMPLE TYPE: C = CHANNEL SAMPLE, R = ROCK CHIP SAMPLE, G = GRAB SAMPLE.

CONCANY, MILTER DECO	IPPES			MIN-FH	1 495	ICP PERAPT				IACT: FTI	PASE I	OF 25SOURCES	
PEOJECT ND: 87 BC 03	0	8	705 WEST	15TH ST., N	IDRTH 1	ANCOUVER, 1	B.C. V7M	112	-	FILE N	0: 7-1413	3/P1+2 030	
ATTENTION: J. WARREN				(604) 980-58	14 DR	(604)988-4	324	+ 1111	E SUIL DEULAEN	T UHIE	78	CP AN-PPR	
(VALUES IN FPH)	85	A5	19	<u>CD</u>	LU	LU					175		
A 601		1	328	3.8	0	.20	1142	1	33	3	105	5 5	
A 602		1	188	3.9	2	17	828	1	20	1	171	4 5	
H 603	2	1	221	3.3	2	17	1042	1	35	1	135	7 10	
A 105		;	200	11	5	20	1745	i	37	4	133	7 5	
A LOL LOW				77		23	997		34	5	134	6 5	
A LAT CTOFAM	7.0	12	790	0 1	8	50	6067	i	181	6	302	1 5	
A LAR LAN		7	303	2.8	i	13	927	1	22	3	110	5 5	Culto
6 LÓD 401		3	313	4.5		26	1476	1	29	5	117	3 5	GWP
A 410	.8	1	215	3.3	6	20	1038	1	31	5	114	5 10	GRID
A 611	.5	i	299	4.2	7	25	1205	1	34	5	121	2 5	
A 617	.7	1	249	3.9	5	19	1609	1	.15	3	118	6 5	9
A 613	.9	7	260	3.6	4	21	934	2	22	4	111	4 5	
A 614 STREAM	4.8	в	588	5.0	6	51	2174	2	121	5	289	3 5	
A 615	.6	1	346	4.1	6	20	1106	1	36	5	113	4 5	
A 616 40M	.1	1	341	3.7	5	17	1248	1	31	2	106	6 5	
A 617	.3	3	317	3.5	5	20	1124	1	35	5	123	.7 10	
A 618 15 cm depl	H .9	26	529	4.3	7	38	1395	1	26	2	120	1 5	
A 619 60 cm II	.4	1	318	3.6	7	33	1239	1	25	5	85	1 5	-
A 520 120 cm 11	.9	6	275	4.8	9	37	1361	1	21	6	95	2 5	TRENCH
A 621 15 cm 11	1.2	1	417	4.4	7	38	1156	1	26	6	102	4 5	4
A 622 GOCM 11	.5	1	353	4.8	В	37	1420	1	24	1	88	1 5	14.10
A 623 /20 cm 11	.8	2	265	4.2	8	34	1259	1	21	1	87	2 5	GWP.
A 624 180 cm 1	4.7	8	385	5.6	9	39	2090	1	85	2	162	1 10	TONE)
A 625 240 CM 1	323.0	121	909	34.4	13	160	12191	12	2471	10	2509	4 630	
A 626 285cm "	126.3	82	692	27.2	13	92	9779	6	1126	1	1795	6 300	
A 627	7.1	1	373	4.7	4	15	2312	1	112	1	247	2 5	
A 628	1.5	4	286	4.6	4	11	2722	1	113	1	183	4 5	
A 529 20H	1.7	12	585	4.1	2	11	2539	2	80	3	202	1 10	
A 630 20H STREAM	1.5	8	448	5.1	. 4	14	2643				170		
A 631	.6	5	348	3.9	1	1	1120	1	63	1	132	1 5	
A 632 40H	.8	12	469	4.9	1	11	3733		/9	:	100	1 10	
A 633	1.6	15	554	4.2	4	10	3064	1	127	2	102	1 10	EAST
A 634	3.5	1	587	8.8	5	14	6496	2	100		3/2	1 15	NIADO
A 635 40M	2.7	2	583	7.3		12	1929		193		240		STOPE
A 636	2.3	5	811	6.5	5	12	9536	4	600	2	211	5 5	VOLE
A 637	1.5	1	447	4.6	2	10	3014	-	100	5	277	2 15	Trave
A 638 STREAM	1.6	1	536	6.1	5	17	3078	-	100	-	147	10 5	YEAK
A 639 40M	2.4	12	847	5.5	8	15	7135		100	1	199	3 20	1000 C 1000
A 640	2.2	13	838	3./	0		2027		10		121	i	
R 642 200	.5	1	160	2.7	-	0	1710		54	1	103	4 5	
R 643 208		1	325	5.2	2	10	3717	1	49	2	192	5 5	
A 011 100	1.7		204	7.5	0	22	6792		141	2	273	12 10	
A 444 40M	2.2	25	771	4.9	5	19	17471		2 105	5	232	1 5	
A 417			170			14	1840		81		227	6 5	
A 449	1.1		2500	5.0	1	21	9237		398	3	269	5 5	
A 449	2.0	1	347	- 7.7	1		1918	2	1 138	1	219	1 5	
4 450	2.0	1	841	2.7	1	9	1036		2 125	4	183	1 5	
A 452	3.2	15	100	6.3	5	26	3137		4 180	1	321	1 10	
4 653	1.1	1	423	3.6	3	8	2732		1 79	1	158	1 5	
A 654 STREAM	1.9	î	546	5.4	4	13	3416	-	1 105	1	249	1 5	- and the second
A 659	79.6	79	715	16.3	18	241	6330	1	5 1205	6	990	1 540	TRENCH
A 550	76.9	72	786	15.2	16	230	5376		4 1125	6	915	1 610	5
A 659	5.8	4	427	3.3	1	16	622		2 302	2	254	1 5	
A 670	1.6	18	539	3.5	3	10	4515		1 50	1	103	1 5	SOUTH
A 671 408	1.5	2	663	4.3	5	21	5330	3	1 . 47	1	149	2 5	WINII
A 672	.4	16	302	3.6	2	4	1757		1 24	1	93	2 5	IVALL
4 673	.4	1	610	3.5	3	2	2433		1 22	4	115	1 5	GRID
	-		150	7.0	2	1	2701		1 71		174	2 4	

	PROJECT NO: 87 1 ATTENTION: J.KA	RESOURCES BC 030 RREN		705 WEST	15TH ST., (604)980-	N LABS IC NORTH VA 5814 OR (P REPORT NCOUVER, 604)958-	B.C. V7K 4524	172 • TYPE 50	IL GEOCH	FILE FILE	1) PAGE ND: 7-141 E:SEPT 29	1 OF 2 150 3/P3+4 03 . 1997 N	IURCES IO	
	(VALUES IN FFM) A6	AS	EA	CD	CG	CU	MN	OK	FP	SB	ZH	CR A	U-FPB	
	A 6/5 40M	9	3	398	3.6	3	B	3938	1	65	1	166	9	5	
	6 477		2	260	3.2	2	10	1863	2	52	1	159		10	
6	A 678 40M	1.7	1	104	4.8	3	13	3463	1	100	2	239	1	5	
	A 677	.7	4	150		5	12	1759	-	70	2	197	2	5	
	A 680	5	2	150	1.0		17	1059				134			
14	A 681	.5	2	244	2.8		17	2245		49	ŝ	174	1	15	
	A 682 20M	.5	6	174	3.7	4	12	2128	i	30	J	154	14	5	
	A 683	.5	5	226	3.1	3	10	1835	i	20	1	146		5	South
	A 684	.7	1	373	3.4	3	11	2891	1	25	i	158	i	10	
	A 685	.5	5	266	3.2	5	13	1294	1	22	1	106	7	5	WALL
	A 686	.4	4	152	3.4	4	12	1008	1	26	1	127	2	5	GRM
	A 687	.6	7	236	3.8	2	9	2379	1	28	2	138	3	5	AUN
	A 688	.5	1	287	3.1	2	9	2285	1	31	1	139	2	10	
	A 699 408	2		196	2.9	3	9	1544	1	17	1	123	1	5	
	A 490	.8	4	168	3.9	5	14	1477	1	31	2	145	8	5	
	H 691	.,	2	140	3.3	4	14	1232	1	18	1	119	7	5	
	H 072	.2	1	94	3.3	1	10	797	1	16	1	116	2	5	
	H 073 400	.2	4	118	3.0	3	11	1127	1	16	1	121	2	10	
	A 105			118	4.0		13	1052	1			113	1	10	
	1010		1	210	4.0	2	14	1101	2	11	1	254	1	5	
-	0 497	1.1	0	317	1.7		16	2291	2	104	1	222		15	
	A 698		ŝ	111	4.7	4		253	1	188	1	105	3	15	
	A 699		2	159	2 2	1	13	70	1	0/	2	126	1	2	00
	A 700 40H		16	544				2047		10		2/			SE
	A 701 40M	1.6	11	465	4.9	1	10	2041	;	145	1	201	1	2	EXTENSIO
	A 702	1.1	11	1287	3.3	2		1779	1	295	i	155	1	10	
	A 703	.9	4	685	3.0	2	5	684	i	207		135	;	15	YELLOW
	A 704	2.1	11	848	6.2	8	12	3383	2	44	4	216	1	10	ROSE
	A 705	5.9	18	369	7.8	5	17	6664	2	418	i	280		15	
	A 706	3.4	3	656	13.8	8	15	4388	1	370	7	608	2	20	FONE
	A 707	2.6	12	498	10.4	8	12	3601	4	295	5	662	1	10	
	A 708	1.0	2	455	4.3	5	12	1946	2	78	1	195	2	10	
	K 101 20H BTR	EAM .6	7	303	3.6	5	16	984	1	24	2	139	5	5	US10754
	K 102 20M STR	EAM .3	1	296	3.4	5	13	1166	1	22	1	130	6	5	
	K 103 201 STR	EAM .8	11	227	5.6	в	24	1544	1	27	4	184	3	5	
	K 104 STR	eam .8	9	307	5.1	7	22	1250	2	20	2	172	11	5	201
	K 105	.3	2	187	3.7	5	21	1194	1	17	7	106	8	15	
	K. 105	.9	4	342	4.7	7	16	2490	1	24	1	144	11	5	
	K 107 408	1.2	10	292	5.1	8	23	1515	- 1	26	10	164	9	5	Neul
	X 109	1.2	1/	339	5.7	8	36	1548	1	26	5	172	6	5	NW
	K 110		14	103	3.1	3	23	1413	5	37	2	49	6	5	EXTENSIO
	r 111		14	323	3.8	•	16	2185	1	24	3	141	9	10	Vellow
	K 112		10	272	5.0		28	1162				248		10	YEHON
	K 113	2.5	147	105	0.6	5	31	2122	17	100	с тт	169	18	13	ROSE
	K 114	2.7	55	219	7.5	10	38	4373	12	174	23	133		5	Tout
	K 115	1.7	18	774	4.0	5	75	1977	2	51	÷	149		5	TONE
	K 116	1.4	14	499	3.9	Å	76	1745	2	17	T	112	7	10	
	K 117	1.9	32	283	6.4	8	37	2457		228	5	400			
	K 118	.9	36	325	5.3	6	27	2483	1	75	4	169	15	5	
	K 119 40M	2.7	53	754	5.8	8	20	7478	3	105	2	158	ę	5	
	K 120	2.3	57	320	5.2	6	21	4425	3	105	ĩ	153	10	5	
	¥ 121 40M	2.7	27	681	5.9	7	16	7869	2	79	1	151	7	5	
	K 122 40M	2.5	26	966	3.5	6	14	5890	2	89	2	105	11	10	
	L 401	.7	1	222	2.6	4	16	1080	1	29	2	82	3	5	1.10
	L 402	.7	4	253	3.4	5	27	1080	1	29	1	62	3	5	GWP
	L 403	.7	3	195	3.1	5	21	1146	1	31	2	74	1	5	GRID
	L 404	.6	4	193	3.1	4	17	956	1	26	2	75	4	15	Aun

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	COMPANY: HI-TEC RESO PROJECT ND: 87 BC 03	URCES 0		705 WEST	HIN-E	N LAPS NORTH	ICP REFOR	T B.C. V7M	172		TJAN	:F31) PAGE FILE NO: 7-1	1 OF 2	SOURCES	
_	ATTENTION: J.WARREN				(604) 980-	5814 OR	(604) 988	-4524	+ TYPE S	DIL GEOCHE		DATE: SEPT 2	1, 1987	£N	
	IVALUES IN FPH 1	AG	AS	BA	CD	CO	CU	-MN	MÐ	F2	SB	ZN	CR	AU-PFB	
	L 405	1.2	4	233	4.1	5	25	1172	1	45	2	122	7	5	
	L 405	1.0	1	173	3.9	4	19	596	2	25	1	108	8	5	GWP
•	L 407	1.4	3	265	4.1	6	27	1036	1	34	3	111	7	10	4
	1 408	.7	в	203	3.9	5	19	767	1	28	1	105	6	5	GRH
	L 409	1.1	5	195	2.7	6	18	870	1	31	3	91	5	15	
	L 410 STREAM	5.1	35	670	6.7	7	43	3816	2	122	3	222	2	10	

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