

ASSESSMENT REPORT
ON THE
ESKAY CREEK PROPERTY
DRILL EXPLORATION PROGRAM
Kay 15 Claim

LOG NO: 0811	RD.
ACTION:	
FILE NO:	

Skeena Mining Division
NTS 104B/9W
Latitude 56°37'N
Longitude 130°29'W
British Columbia

FILMED

July 14, 1989

by

D.W. Mallo
Prime Explorations Ltd.

18,958

GEOLOGICAL BRANCH
ASSESSMENT REPORT

Part 1 of 2

Owner:

CONSOLIDATED STIKINE SILVER LTD.
800-900 West Hastings Street
Vancouver, British Columbia
V6C 1E5

Operator:

CALPINE RESOURCES INCORPORATED
PRIME CAPITAL PLACE
10th Floor,
808 West Hastings Street
Vancouver, British Columbia
V6C 2X6

ESKAY CREEK PROJECT

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INTRODUCTION

Objective

The objective of the 1989 drill programme on the Eskay Creek Project was to test the property for gold and silver mineralization at depth. This report presents data on four holes that were drilled for a total of 838.4 metres on the Kay 15 claim.

Location and Access

The Eskay Creek project area is located 83 kilometers (52 miles) north-northwest of Stewart, B.C. (Figure 1) and 37 kilometers (23 miles) east of the Prime Resources Corporation and Cominco Limited SNIP Deposit. Access is by helicopter from Bronson, Stewart or Bell II on the Stewart-Cassiar Highway, 25 kilometers east. Tom MacKay Lake, 5 kilometers to the west is suitable for float plane landings, and an unused track runs from the property to the lake shore. An abandoned short airstrip is situated 10 kilometers south of the property.

Road access within the region is currently under review, and several different development corridors are proposed. One such corridor would pass within 20 kilometers of the project area.

Physiography

The Eskay Creek Project is located on the Prout Plateau within the eastern flank of the Coast Mountains. The Prout Plateau is a rolling massif ranging from 850 to 1300 meters elevation above sea level, characterized by severely glaciated, rocky terrain and subalpine vegetation. Relief on the property is approximately 200 meters and is locally sharp.

The property straddles a ridge, with Argillite Creek on the west and Eskay Creek on the east. Both creeks drain north and join Mackay Creek, a tributary of the south-flowing Unuk River. The Unuk River valley is located approximately 2.5 kilometers east of the property, comprising a relatively narrow, heavily forested canyon.

Climate

There is no meteorological data for the immediate project area. Historical records and current operating experience suggest that annual precipitation exceeds 300 mm, much of which falls as snow. Summer conditions last from late June to the end of September, and are characteristically coast insular or temperate and wet. Winter conditions span the remaining calendar months, during which snow accumulations can exceed 10 meters. Winter conditions are difficult to predict, as the controlling factor is a

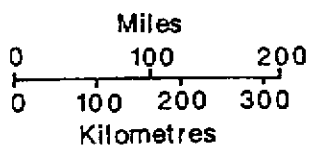


PRIME EXPLORATIONS LTD.

CALPINE RESOURCES INC.

ESKAY CREEK PROJECT

LOCATION MAP



Date: Oct. 1988

Drafted by: E.R.

FIGURE 1

continuous onslaught of warm, moist low pressure systems from the Gulf of Alaska which rise over the Coast Range and dump snow along the divide regions. The interior region immediately east is usually dominated by an arctic high pressure cell, which can become extremely well-entrenched, resulting in outflow conditions which can see both prolonged spells of cold, dry weather and high wind conditions.

Notwithstanding the unpredictable nature of the local weather, year long operations can be sustained by maintaining a properly winterized camp and providing a programme of avalanche control. Numerous other year round development projects are underway within the region, and the current exploration programme at Eskay has clearly demonstrated that cost-effective winter exploration campaigns can be mounted.

Claims

The Eskay Creek Property consists of thirty 2-post mineral claims located in the Skeena Mining Division (Figure 2). The claims are situated in NTS map-sheet 104B/9W, centered about 56°37' north latitude, and 130°29' west longitude. Claim descriptions are as follows:

<u>Claim Name</u>	<u>Record Nos.</u>	<u>Location Date</u>	<u>Expiry Date</u>
TOK 1-6	37248-37253	May 25, 1972	May 31, 1999
TOK 7-14	37254-37261	May 26, 1972	May 31, 1999
TOK 15-22	37421-37428	Aug. 16, 1972	Sep. 6, 1999
KAY 11-18	21077-21084	Oct. 2, 1962	Oct. 11, 1999

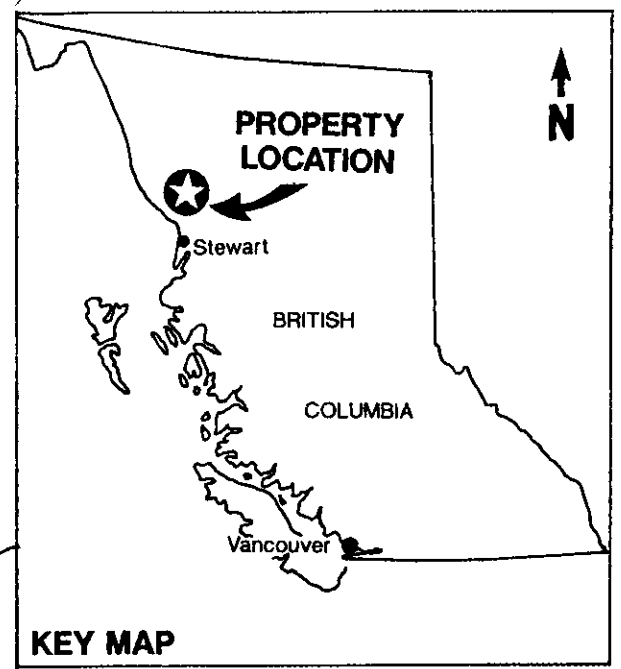
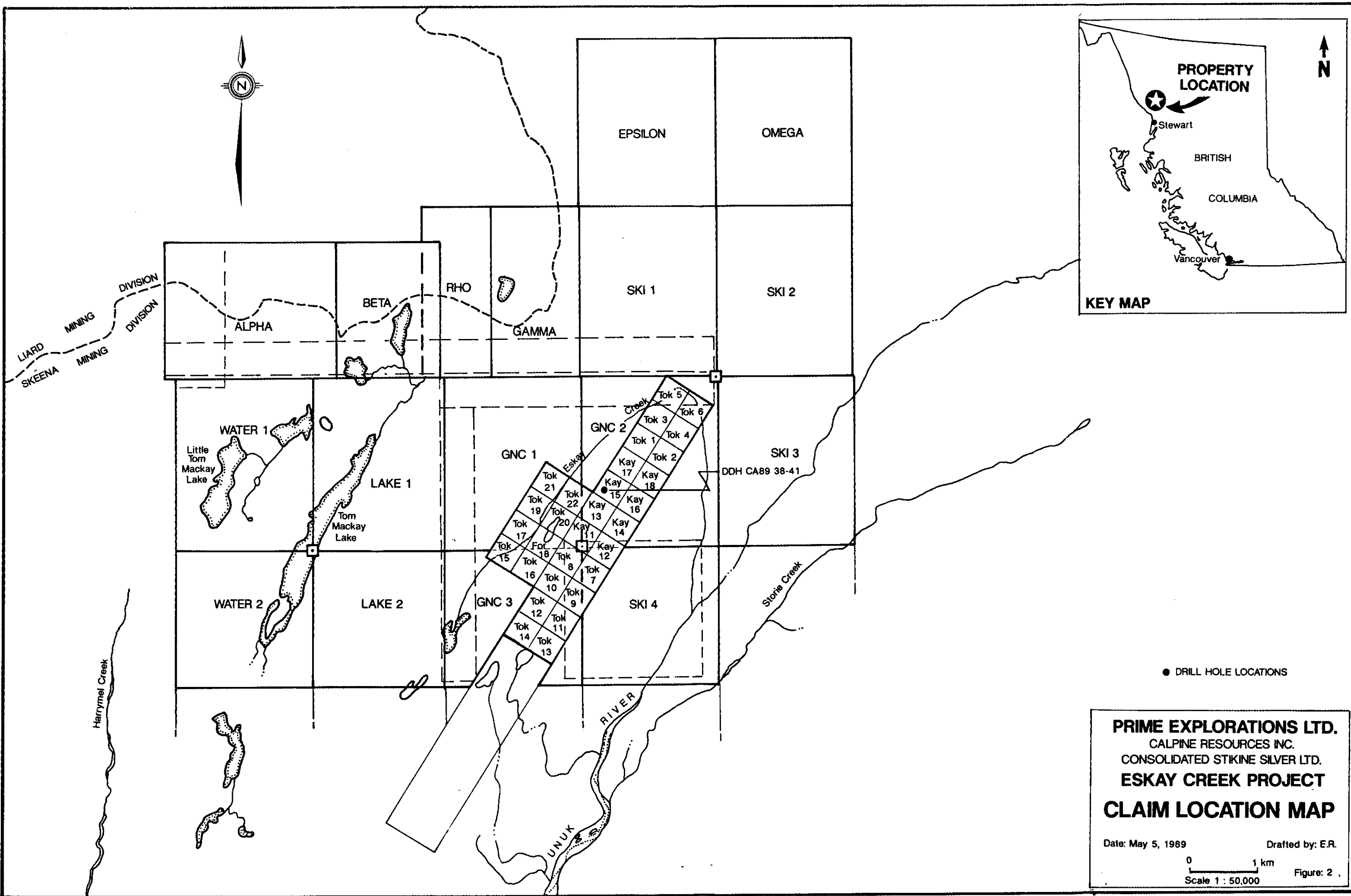
History

The Eskay Creek area has undergone numerous exploration campaigns since discovery in 1932. The property is now operated on a joint venture basis between Calpine Resources Incorporated and Consolidated Stikine Silver Limited, with Prime Explorations Ltd. as the Project Manager.

Drilling

A winter drill exploration programme was initiated in mid-January of 1989 and completed in early May comprising 13,467.9 metres in 54 holes of NQ core. The drilling was done under contract by Falcon Drilling Ltd, of Prince George, British Columbia. All core was logged, split and stored on site at the Calpine camp.

Drill holes CA89-38 to 41 inclusive are submitted for this assessment report. Each drill hole has a length of approximately 200 metres with a lithology consisting of interlayered andesites and black argillites in the top part of each hole, and rhyolites and felsic volcanics bottoming the holes.



● DRILL HOLE LOCATIONS

PRIME EXPLORATIONS LTD.
 CALPINE RESOURCES INC.
 CONSOLIDATED STIKINE SILVER LTD.
ESKAY CREEK PROJECT
CLAIM LOCATION MAP

Date: May 5, 1989 Drafted by: E.R.

0 1 km
 Scale 1 : 50,000 Figure: 2

The split core samples were sent to Bondar-Clegg Laboratories in North Vancouver for gold and silver assays. Selected pulps were sent to TSL Laboratories, Saskatoon, for check analysis. Core samples sent to Bondar Clegg were fire assayed with an AA finish for Au and Ag using a one assay ton sample size. If the gold value was >0.100 ounce per ton, then the sample was re-assayed with a gravimetric finish. Samples with >.750 ounce per ton Au were analyzed for metallic gold (metallic sieve assay). Analytical procedures are shown in Appendix I.

Core samples sent to TSL Labs were fire assayed with a gravimetric finish for gold. Silver was assayed using an acid digestion (HCL-HNO3) with an AA finish.

Core logging and splitting was done under the supervision of G. McArthur. A list of contractors is shown in Appendix II and statement of qualifications is shown in Appendix III. The drill logs, assays and drill hole plan map and Au cross section for holes CA89-38 to 41 are shown in Appendix IV.

The statement of Expenditures is in Appendix V.

GEOLOGY

Regional Geology

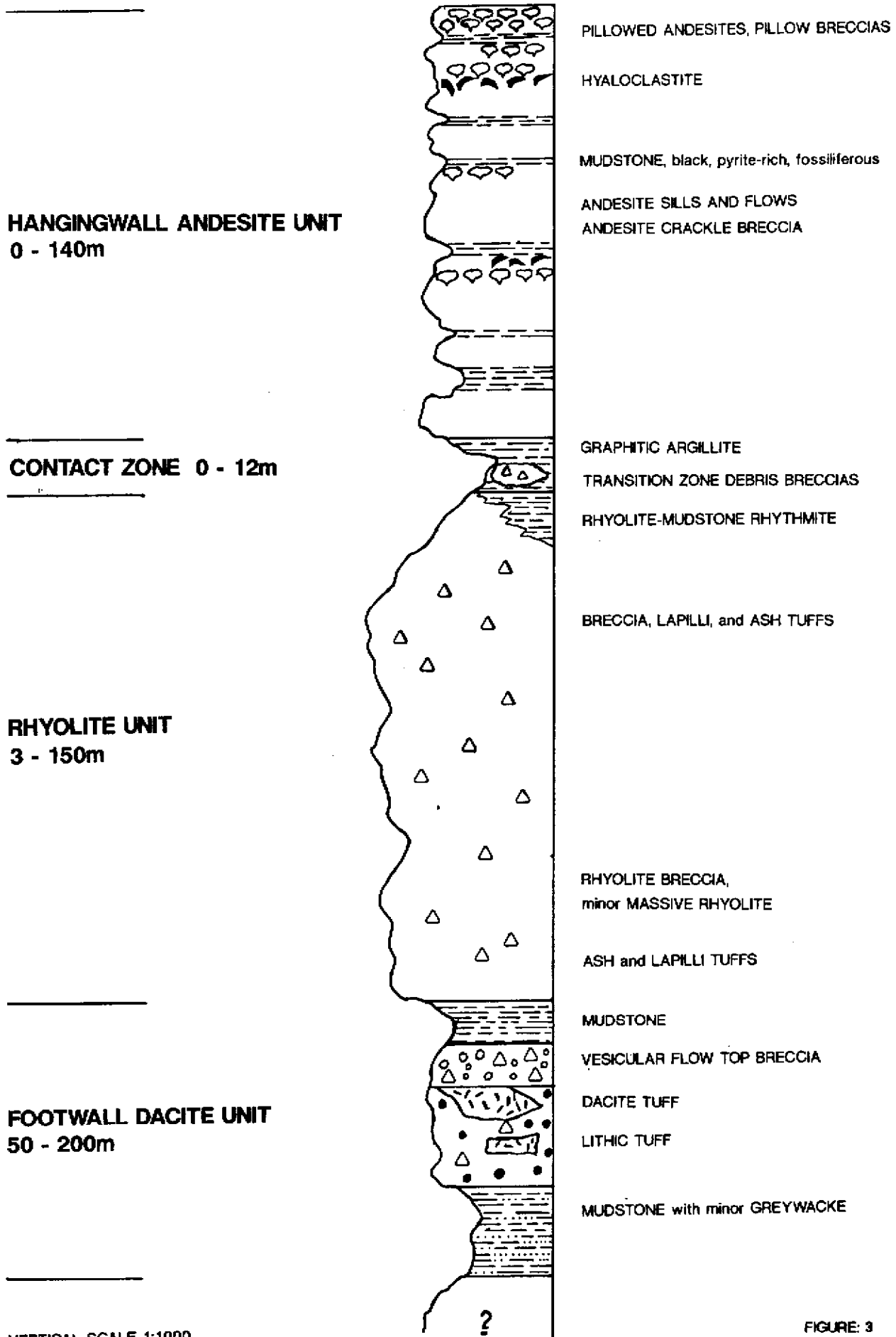
The Eskay Creek area lies within the Intermontane tectonic belt, containing Stikine terrane rock assemblages. The Unuk River area is underlain by a thick, weakly metamorphosed Upper Triassic to Lower Jurassic volcanic and sedimentary arc-related units overlain by Middle Jurassic successor basin sedimentary units (Britton et al, 1989). Inconclusive evidence of late Triassic deformation exists. Large-scale northeast plunging vertical folds and major north-trending cataclastic and fault zones are thought to be principally related to late Jurassic to early Cretaceous plutonism and orogenesis.

Regional geological mapping by the Geological Survey of Canada, the British Columbia Ministry of Energy, Mines and Petroleum Resources and Newmont Mining Corp. (Granduc Mines Ltd.) has resulted in selective areal map coverage and a working stratigraphic column (Figure 3). Government reconnaissance mapping is on-going, and revision to the current understanding is anticipated. Rock unit correlation is based upon fossil control and gross unit similarities to adjacent southern map areas where more detailed geological mapping has been undertaken.

Stuhini Group

Upper Triassic volcanic and sedimentary rocks tentatively correlated to the Stuhini Group occur east of the Unuk River and

STRATIGRAPHIC COLUMN - 21 ZONE



VERTICAL SCALE 1:1000

FIGURE: 3

west of Harrymel Creek. Stuhini rocks include variably deformed and metamorphosed siltstone, wacke, conglomerate and limestone overlain by basalt to andesite flows and breccias and locally dacite pyroclastic tuffs and breccias.

Hazelton Group

Unuk River Formation:

Earliest Lower Jurassic Unuk River Formation occur at moderate elevations east of the Unuk River and west of Harrymel Creek. The Unuk comprises a relatively monotonous sequence dominated by green andesite tuffs, flows and subordinate pyroclastic rocks, intercalated with wacke, siltstone, and minor conglomerate.

Betty Creek Formation:

Overlying the Unuk is the Lower Jurassic Betty Creek Formation, outcropping throughout the Unuk valley. The Betty Creek comprises red, maroon to green volcanoclastic conglomerate, andesite and dacite pyroclastic tuff and breccias with intercalated grit and arenaceous wacke.

Mount Dilworth Formation:

Overlying the Betty Creek is the Lower Jurassic Mount Dilworth Formation, outcropping on the Prout Plateau and at higher elevations west of Harrymel Creek and east of the Unuk River. The Mount Dilworth comprises dacite to rhyolite pyroclastic breccias, bedded tuff and subordinate flows and flow breccias.

Salmon River Formation:

Late Lower Jurassic Salmon River Formation outcrops north and west of the Prout Plateau. It comprises a drab sequence of grey siltstone, fine-grained arenite, chert and limestone.

Bowser Group

Ashman Formation:

Middle Jurassic units thought to be equivalent to the basal Ashman Formation occur on the Prout Plateau in the vicinity of Tom Mackay Lake. Ashman rocks include chert pebble conglomerate, grey to black mudstone and wacke and subordinate limestone and mafic volcanic flows.

Cenozoic to Recent subaerial olivine basalt flows and tephra are distributed widely in the region, though none are reported on the Prout Plateau. Deposits are widespread in the major river valleys, such as the Unuk, as well as in the Cone Glacier area, west of Harrymel Creek. Valley bottom deposits tend to be characterized by palisade-type sheet flows. At higher elevations ice-contact cones, domes and tephra fields predominate. Numerous felsic and mafic dykes, thought to be coeval with the young volcanic deposits, are locally abundant.

APPENDIX I

Analytical Procedures - Bondar-Clegg Ltd., Vancouver

PROCEDURE FOR ASSAY Au ANALYSIS

Fire Assay Procedure:

A prepared sample of one assay ton (29.166 grams) is mixed with a flux which is composed mainly of lead oxide. The proportions of the flux components (the litharge, soda, silica, borax glass and flour) are adjusted depending upon the nature of the sample. Silver is added to help collect the gold. The samples are fused at 1950°F until a clear melt is obtained. The 30-40 gram lead button that is produced contains the precious metals. It is then separated from the slag. Heating in the cupellation furnace separates the lead from the noble metals. The normal-sized precious metal beads that are produced are transferred to test tubes and dissolved with aqua-regia. This solution is analyzed using Atomic Absorption by comparing the absorbance of these solutions with that of standard solutions. In case of high grade samples, the precious metal bead is parted to separate the silver and the remaining gold is weighed.

Comments:

As part of routine quality control duplicate analyses are done for about 15% of the samples. Also, all samples which are over 0.20 ounce per ton on the original fusion are run again to verify the results. If a sample gives erratic results, such as 0.10, 0.020, 0.30, these are indicated on the report. It is suggested that a new split should be taken from the reject for preparation and analysis by metallics sieve procedure. These assay results will always be signed by the registered assayer.

PROCEDURE FOR ASSAY Cu, Pb, Zn, AND Ag BY ATOMIC ABSORPTION ANALYSIS

Samples of 0.5 grams are weighed with 0.25 gram duplicates and digested in glass beakers with concentrated nitric and hydrochloric acids. The beakers are heated on the hot plate until the solution completely dries, and then the samples are redissolved with dilute hydrochloric acid. The solutions are run by Atomic Absorption, using the appropriate lamp and wavelength for each element. The absorbency for each element is recorded and compared to a standard series to determine the amount present.

The procedure is similar for assay Ag by Atomic Absorption, except the sample weight is 3 grams and hydrofluoric acid is also added during the digestion. Background correction is introduced in analyzing Ag on the A.A. to overcome the matrix problem.

Comments:

All samples having a value greater than 20% Cu or Pb, 10% Zn, or 10 ounces per ton Ag have to be rerun by classical methods.

APPENDIX I
Analytical Procedures

BIBLIOGRAPHY

- Aldrick, D.J., Britton, J.M., Webster, I.C.L. and Russell, C.W.P., (1989); Geology and Mineral Deposits of the Unuk Area, B.C. GSB Open File Map 1989-10
- Annual Reports of the Minister of Mines, British Columbia;
1933, p 61; 1934, pp B30-B33; 1935, pp B9, B27; 1939, p 65;
1946, p 85; 1953, pp 87-89; 1963, p 10; 1964, p 20;
1965, p 44; 1967, p 30
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(1970), pp 64-65; and (1971) p 36
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- Grove, E.W. (1971); Geology and Mineral Deposits of the Stewart area, Northwestern British Columbia Department of Energy, Mines and Petroleum Resources, Bulletin No. 58
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Creek Fault). The major structure on the property appears to be a shallow northeast plunging asymmetric anticline. Fold limbs are either faulted and/or steeply dipping. The Eskay Creek Property is characterized by highly altered and gossanous rhyolitic units which host several gold-silver and minor base metal occurrences.

Gold mineralization is hosted within both rhyolite breccia and graphitic mudstone formations, beneath barren andesite flows. Disseminated stibnite is ubiquitous to the mudstone unit, developing into high grade massive stibnite-realgar bands up to 20 feet thick at the mudstone-rhyolite contact. Footwall alteration is intense, comprising sericite and gypsum in highly sheared rhyolite tuff and breccia. Associated sulphide minerals include stibnite, arsenopyrite and pyrite.

Steeply dipping faults and associated fractures appear to be a major controlling feature of the mineralization. Mechanically induced fracture systems formed channels which divided the mineralizing hydrothermal fluids. This produced the pattern of high grade ore zones surrounded by overlapping lower grade zones. Secondary control of gold mineralization was provided by host rock lithology.

RECOMMENDATION

Additional drilling needs to be done in order to delineate the extent of gold/silver mineralization. This should be done in conjunction with a summer exploration programme consisting of geological mapping, prospecting, and geophysical surveys to define additional drill targets.

- (1) Andesite flows, breccia and tuff with interbedded wacke and siltstone,
- (2) Tuffaceous wacke, mudstone, and conglomerate,
- (3) Dacite lapilli, crystal and lithic tuffs interbedded with black mudstone and waterlain tuff,
- (4) Rhyolite lapilli tuff and breccia,
- (5) Pillowed andesite flows and breccias with interbedded carbonaceous mudstone, and
- (6) Medium to thin-bedded conglomerate, wacke and mudstone.

Well preserved micro and macrofossils are locally abundant in most sedimentary units, providing relatively rigid stratigraphic control and indicating a predominately subaqueous depositional environment. Units 1 to 3 are tentatively correlated to the Betty Creek Formation, unit 4 to the Mt. Dillworth and units 5 and 6 to the Salmon River and/or Ashman Formations. Stratigraphic assignment may change pending the results of additional government mapping planned for 1989.

The major structure on the property appears to be a shallow northeast plunging asymmetric anticline with a steep eastern limb. The western limb is cut by the major Argillite Creek fault. The fold closes across the northernmost portion of the property at Mackay Creek, and it appears likely that favourable rhyolite geology may be preserved at depth north and east of the #23 showing area.

Penetrative cleavage is observed in sedimentary rocks and the upper portion of the rhyolite, striking 030/75-80W. All other units are relatively undeformed. Metamorphic rank is sub-greenschist. A northeast-trending vertical fault of unknown displacement separates rocks of units 5 and 6 (Argillite Creek fault). A similar, parallel structure is postulated to underlie Eskay Creek upstream from the camp. Major and minor north and east-trending faults of unknown attitude and displacement have been mapped within units 3, 4 and 5. Numerous airphoto lineaments are suspected to represent faults, though some may be joints. Plans by Whiting portray many short-length faults, some of which coincide with airphoto features. Descriptions of the Northend prospect report flat-lying, small-displacement faults occupied by barren quartz veins. These are reminiscent of minor structures reported at the Snip and Johnny Mountain Deposits to the north in the Bronson Creek area, and may reflect the presence of low-angle reverse faults and/or post-mineral extension fissures.

CONCLUSIONS

The Eskay Creek Property is underlain by Lower to Middle Jurassic volcanic and sedimentary rocks of the Hazelton Group separated from the nearby Triassic Bowser Group of marine-basin sediments by a possibly deep-seated fault structure (Argillite

Intrusive Rocks

Government mapping has not located any intrusive rocks on the Prout Plateau. Elsewhere in the region a variety of intrusives are documented, including Triassic gneissic quartz diorite stocks, Jurassic diorite and gabbro stocks and feldspar-porphyrific granodiorite and syenite stocks and sills, and Tertiary feldspar-porphyrific monzonite stocks and felsic or basic dyke swarms. The eastern contact of the Tertiary Coast Plutonic Complex is approximately 25 kilometers southwest of Eskay Creek.

Metamorphism

According to Britton et al (1989) regional metamorphic rank is lower greenschist, characterized by saussuritized plagioclase feldspar, chlorite after mafic minerals, and white mica after clay. Metamorphic rank locally increases to lower amphibolite within one kilometre of the Coast Plutonic Complex. Contact metamorphic hornfels zones are common adjacent to the larger igneous intrusives.

Deformation

Folding:

Outcrop to regional scale, upright to slightly overturned vertical folds are documented both in the Eskay Creek area and the surrounding region. Fold axes trend 020 to 035° North, plunging 0 to 15°N. On the Prout Plateau a schistose rock fabric is present which may reflect this phase of deformation.

Faulting:

Topographic lineaments are abundant in the area, and many likely reflect faults or joints. Documented structures are rare, including small displacement normal and reverse faults. A major 150°N-trending schistose shear zone occupies the lower Unuk River valley, which to the north bifurcates or joins a major north-trending mylonite and cataclasite band underneath the Harrymel Creek valley and a major vertical fault under Clouter and Argillite Creeks on the Prout Plateau. Recent movement on the Harrymel structure is normal, however the zone is postulated to be an older, deep-seated major fault zone of unknown displacement.

Property Geology

The Eskay Creek Property is underlain by Lower to Middle Jurassic volcanic and sedimentary rocks of the Hazelton Group. Rock units are west-facing, striking 060°N/15-70.°W. Dips are steepest in the central and southern portion of the property, and become more shallow to the north. From oldest to youngest units, the stratigraphic section includes:

APPENDIX II
Contractor Services

APPENDIX II

Contractor Services

CONTRACTOR

Central Mountain Air Ltd.
P.O. Box 998
Smithers, British Columbia
VOJ 2N0

Transportation -
Fixed Wing

Falcon Drilling
1901 Olgilvie Street
Prince George, British Columbia

Drilling

Jaycox Industries
P.O. Box 3633
Smithers, British Columbia
VOJ 2N0

Expediting

Northern Mountain Air Ltd.
P.O. Box 368
Prince George, British Columbia
V2L 4S2

Transportation -
Helicopter

APPENDIX III

Certificate of Qualifications

APPENDIX III

Certificate of Qualifications

I, David W. Mallo of 4775 Hermatige Drive, Vancouver, British Columbia hereby certify:

1. I am a graduate of Brandon University (1981) and hold a BSc (Spec) degree in geology.
2. I have been employed in my profession by various mining companies since graduation.
3. I am presently employed as a senior geologis with Prime Explorations Ltd., of 1000-808 West Hastings Street, Vancouver, British Columbia.



David W. Mallo
Senior Geologist

DATED at Vancouver, British Columbia, this 14th day of July, 1989.

CERTIFICATE OF QUALIFICATIONS

I, Gerald F. McArthur of Delta, British Columbia hereby certify:

1. I am a geologist with a business address at 11135 Monroe Drive, Delta, British Columbia, V4C 7T2.
2. I am a graduate of the University of British Columbia with a BSc geology (1973).
3. I have practised my profession in mineral exploration since graduation.
4. I am a Fellow of the Geological Association of Canada and a Professional Geologist registered in the Province of Alberta.



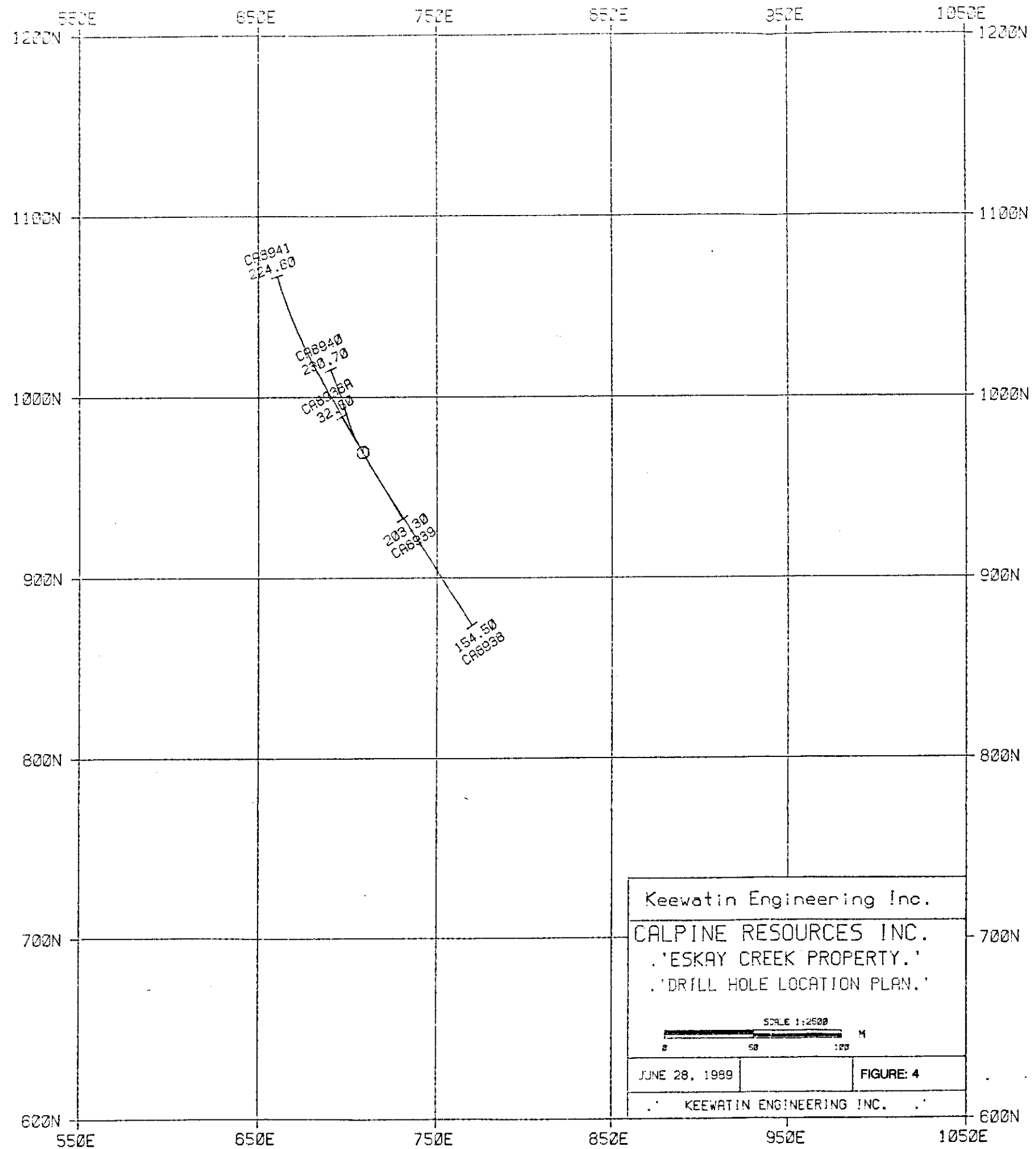
Gerald F. McArthur
Geologist

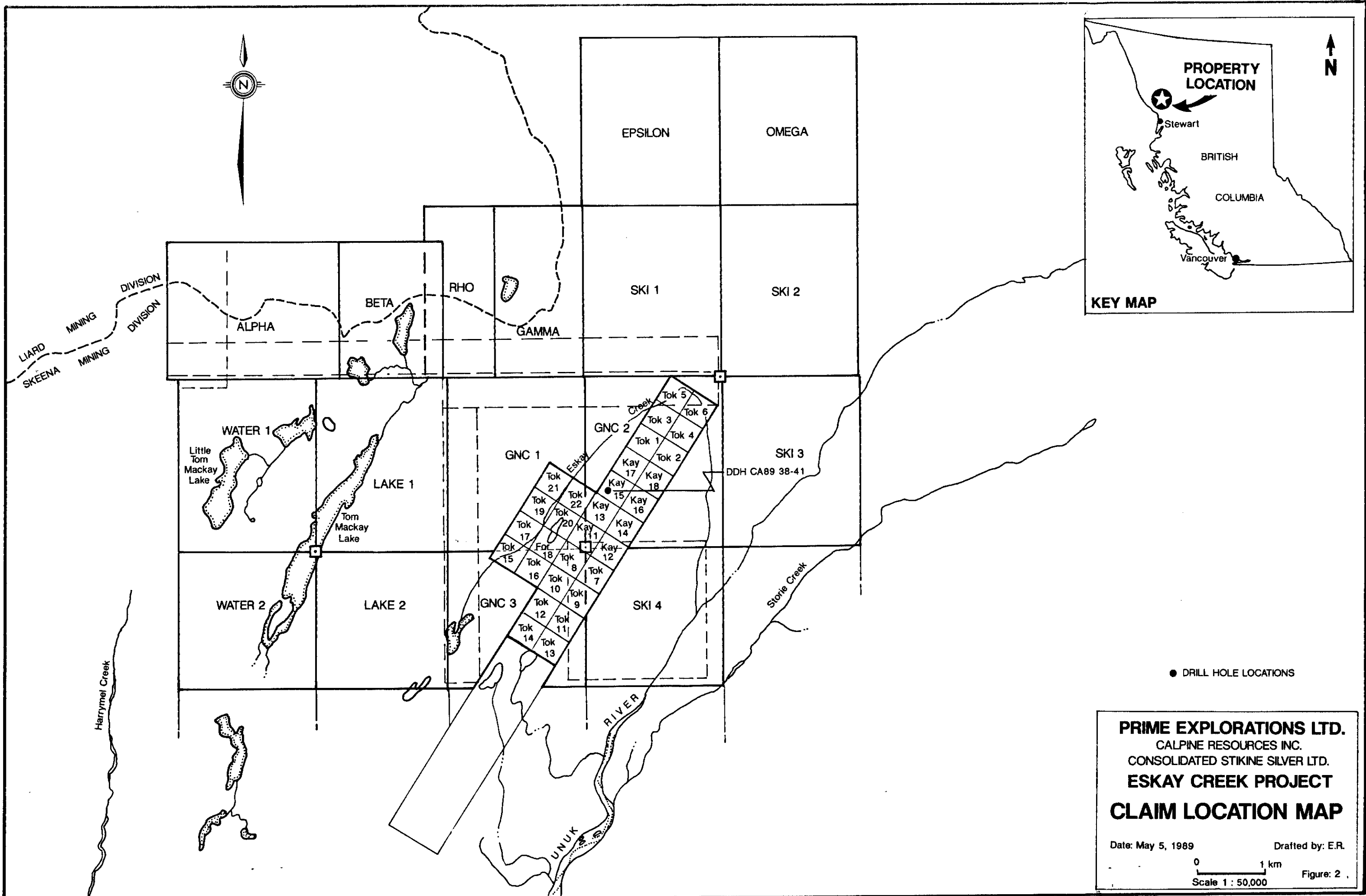
DATED at Vancouver, British Columbia, this 26th day of *June*, 1989

APPENDIX IV

Diamond Drill Summary

Diamond Drill Logs and Assays
Drill-Au Section 1:1000





● DRILL HOLE LOCATIONS

PRIME EXPLORATIONS LTD.
 CALPINE RESOURCES INC.
 CONSOLIDATED STIKINE SILVER LTD.
ESKAY CREEK PROJECT
CLAIM LOCATION MAP

Date: May 5, 1989
 Drafted by: E.R.

0 1 km
 Scale 1 : 50,000
 Figure: 2

DRILL HOLE LOG

LOCATION: 969.451N 708.697E

HOLE NO.: CA-89-38 PAGE NO: 1

AZIM.: 150 ELEV.: 99.1 m
 DIP: 45° LENGTH: 154.4 m
 CORE LENGTH: NQ

DIP TEST
 Meterage Dip Azimuth
 48.8 m -42° 147°

STARTED: March 7-8/89
 COMPLETED: March 10/89 - lost hole
 PURPOSE: test zone upper tier holes

PROPERTY:
 CLAIM NO.:
 SECTION: 2+50S
 LOGGED BY: GF. McArthur
 DATED LOGGED: March 9/89
 DRILLING CO.: Falcon
 ASSAYED BY: Bondar Clegg

CORE RECOVERY: Lost hole in fault unable to reenter for survey below 50 m

From	To	Description	Sample No.	From	To	Length	Au ppb	Au oz/t	Ag ppm	Ag oz/t	Pb ppm	Cu ppm	Zn ppm	As ppm	Sb ppm
0	4.0	HANGING WALL ANDESITE OB CASING - no recovery													
14.0	78.3	UPPER ANDESITE	109350	5.3	9.5	4.2		<0.002		<0.02					
		- andesite fine grained brown-greenish	109351	3.8	5.3	1.5		<0.002		<0.02					
		- white calcite veining	109352	10.5	11.1	0.6		<0.002		<0.02					
		- crackle breccia filling	109353	14.8	16.3	1.5		<0.002		<0.02					
		- several (4) calcite events different color white (black-dol?)	109354	23.5	25.0	1.5		<0.002		<0.02					
		- interbedded with black argillite	109355	26.0	27.5	1.5		<0.002		<0.02					
		- pyrite lam beds	109356	30.8	31.3	0.5		<0.002		<0.02					
		4.0 - 4.7	109357	36.0	38.0	2.0		<0.002		<0.02					
		- crackle breccia with internal white calcite remmed by xline black-brown calcite vein 20° to core axis	109358	40.5	42.0	1.5		<0.002		<0.02					
			109359	42.0	43.5	1.5		<0.002		<0.02					
			109360	43.5	45.0	1.5		<0.002		<0.02					
			109361	45.0	46.5	1.5		<0.002		<0.02					
		7.2 - 7.	109362	46.5	48.0	1.5		<0.002		<0.02					
		- calcareous vein breccia	109363	48.0	49.5	1.5		0.002		<0.02					
		7.4 7.5	109364	49.5	51.0	1.5		0.002		<0.02					
		- calcite fracture 15 - 20° to core axis	109365	51.0	52.5	1.5		<0.002		<0.02					
			109366	52.5	54.0	1.5		0.002		<0.02					
		8.2	109367	54.0	55.5	1.5		<0.002		<0.02					
		- grey calcite breccia	109368	55.5	57.0	1.5		<0.002		<0.02					
		- oxidized rubble	109369	57.0	58.5	1.5		<0.002		<0.02					

From	To	Description	Sample No.	Sample			ASSAYS							
				From	To	Length	Au ppb	Au oz/t	Ag ppm	Ag oz/t	Pb ppm	Cu ppm	Zn ppm	As ppm
		78.2 - 45° to core axis - pyritic layers 60° to core axis - calcite veining 15° to core axis - base of argillite is cherty												
78.8	154.4	FELSIC VOLCANIC UNIT - fine to coarse ash tuffs - large flow banded or fine siliceous flow - fragments 60 cm range - some glassy sections - possible coarse to fining upward	109383	78.5	80.0	1.5		<0.002		<0.02				
78.3	79.5	- siliceous greyish rhyolite fine ash tuff 1 - 2 cm fragments - oxidized - broken ground, fault												
79.5	84.0	- grey matrix supported, banded fragments 2 - 4 cm - light greenish harder fragments - blocky subrounded	109384 109385 109386	80.0 81.5 83.0	81.5 83.0 84.5	1.5 1.5 1.5		<0.002 0.004 0.003		0.03 0.05 0.05				
84.0	84.4	- greenish matrix light green clasts matrix soft - fine grained 1 - 2 cm fragments												
84.4	87.0	- coarse grained 2 - 5 cm fragments	109387 109388	84.5 86.0	86.0 87.5	1.5 1.5		0.003 0.003		0.06 0.14				
37.0	87.4	- matrix supported 70% matrix												
37.4	87.5	- fine ash, 45° to core axis	109389	87.5	89.0	1.5		0.008		0.10				
37.5	91.7	- ash tuff, matrix supported 89.5 - 9 - closely packed fragments	109390	89.0	90.5	1.5		0.010		1.75				

From	To	Description	Sample No.	Sample		Length	Au ppb	Au oz/t	Ag ppm	ASSAYS					
				From	To					Ag oz/t	Pb ppm	Cu ppm	Zn ppm	As ppm	Sb ppm
91.7	92.0	- fine ash, 50° to core axis	109391	90.5	92.0	1.5		0.006							
			109392	92.0	93.5	1.5		0.003							
92.0	95.7	- fragments 2 - 4 cm greenish	109393	93.5	95.0	1.5		0.006							
		95.7 - 93	109394	95.0	96.5	1.5		0.002							
		- blotchy light green calcareous alteration	109395	96.5	98.0	1.5		0.003							
			109396	98.0	99.5	1.5		0.004							
		93 - 95.6	109397	99.5	101.0	1.5		0.002							
		- ash tuff, fragments 2 cm	109398	101.0	102.5	1.5		0.002							
		95.6 - 96.6	109399	102.5	104.0	1.5		<0.002							
		- coarsening, fragments to 6 cm grey	109400	104.0	105.5	1.5		0.002							
		96.6 - 97.5	109401	105.5	107.0	1.5		<0.002							
		- finer grained ash, bedding 45° to core axis	109402	107.0	108.5	1.5		0.002							
			109403	108.5	110.0	1.5		0.002							
		97.5 - 99.5	109404	110.0	111.5	1.5		0.005							
		- coarser ash tuff, 1 - 2 cm fragments	109405	111.5	113.0	1.5		0.015							
		99.5 - 99.7	109406	113.0	114.5	1.5		0.002							
		- matrix dominant dark grey	109407	114.5	116.0	1.5		0.003							
		99.7 - 100	109408	116.0	117.5	1.5		0.008							
		- coarser ash, more fragments, 1 - 2 cm	109409	117.5	119.0	1.5		0.003							
		100 - 100.1	109410	119.0	120.5	1.5		0.002							
		- fine ash, 45° to core axis	109411	120.5	122.0	1.5		<0.002							
		100.1 - 101	109412	122.0	123.5	1.5		0.002							
		- ash tuff, - 1 - 2 cm fragments	109413	123.5	125.0	1.5		<0.002							
		101 - 102	109414	125.0	126.5	1.5		<0.002							
		- matrix dominant ash tuff	109415	126.5	128.0	1.5		<0.002							
		102 - 103	109416	128.0	129.5	1.5		<0.002							
		- ash tuff	109417	129.5	131.0	1.5		<0.002							
		103 - 105	109418	131.0	132.5	1.5		<0.002							
		- fine grained section	109419	132.5	134.0	1.5		0.002							
		105 - 111	109420	134.0	135.5	1.5		0.002							
		- ash tuff, 2 - 6 cm fragments	109421	135.5	138.0	2.5		0.002							
		111 - 124	109422	138.0	139.5	1.5		0.002							
		- fine ash, 1 cm or less, occasional	109423	139.5	141.0	1.5		0.003							

From	To	Description	Sample No.	From	To	Length	Au ppb	Au oz/t	ASSAYS				
									Ag ppm	Ag oz/t	Pb ppm	Cu ppm	Zn ppm

- trace pyrrhotite
 - calcite fracture filling throughout
- 187.4 to 203.8
- fractures from 50° to 70° to core axis
 - up to 1 cm wide
 - frequency of 1 per 20 cm

END OF HOLE

DRILL HOLE LOG

LOCATION: 969.451N 708.697E

HOLE NO.: CA-89-40 PAGE NO: 1

AZIM.: ELEV.: 99.1 m
 DIP: LENGTH: 230.7 m
 CORE LENGTH: NQ

PROPERTY: Eskay Creek

STARTED: March 14/89
 COMPLETED:
 PURPOSE:

DIP TEST
 Meterage Dip Azimuth
 115 m -78° 344°
 231 m -76.7° 337°

CLAIM NO.:
 SECTION: 2+50S
 LOGGED BY: Christine Swanson
 DATED LOGGED: March 16/89
 DRILLING CO.: Falcon
 ASSAYED BY: Bondar Clegg

CORE RECOVERY:

From	To	Description	Sample No.	From	To	Length	Au ppb	Au oz/t	Ag ppm	ASSAYS Ag oz/t	Pb ppm	Cu ppm	Zn ppm	As ppm	Sb ppm
0	3.0	OVERBURDEN AND CASING													
3.0	23.6	ANDESITE FLOWS	109549	7.5	9.0	1.5		<0.002		<0.02					
		- pale green	109550	15.2	16.6	1.4		<0.002		<0.02					
		- aphanitic to fine grained phaneritic	109551	20.0	22.0	2.0		<0.002		<0.02					
		- locally amygdular													
		- local crackle breccia with matrix of grey calcite, 1% pyrite, 1% pyrrhotite, trace chlorite													
		- occasional calcite fracture filling from 35° to 55° to core axis													
		- up to 1 cm wide													
		- frequency of 1 per 30 cm													
		- rusted, intensely fractured zone from 15.2 to 17.1 m													
		21.5 - 22.8													
		- calcareous argillite rubble from 21.5 - 22.6 massive from 22.6 to 22.8													
23.6	33.1	ARGILLITE	109552	24.5	26.0	1.5		<0.002		0.02					
		- medium grey calcareous layers up to 30 cm thick interlayered with dark black argillite and lighter coarser	109553	27.5	29.0	1.5		<0.002		0.04					
			109554	32.0	33.5	1.5		<0.002		0.02					

From	To	Description	Sample No.	From	To	Length	Au ppb	Au oz/t	Ag ppm	ASSAYS				
										Ag oz/t	Pb ppm	Cu ppm	Zn ppm	As ppm
		- very siliceous	108567	78.0	79.5	1.5		<0.002		<0.02				
		- clasts ranging from coarse ash to 7 cm across	108568	79.5	81.0	1.5		<0.002		<0.02				
		- pale grey clasts in a darker matrix	108569	81.0	82.5	1.5		<0.002		<0.02				
		- 10% matrix	108570	82.5	84.0	1.5		0.003		<0.02				
		- 1 - 3% pyrite as fine grained wispy portions of matrix and as occasional blebs up to 1 cm across	108571	84.0	85.5	1.5		<0.002		<0.02				
		- high fracture density, 1 per 3 cm												
		- lower contact at 30° to core axis												
		- definite contact												
		- moderately foliated, 35° to core axis from 85.0 - 85.4												
85.4	91.8	GRAPHITIC ARGILLITE	108572	85.5	87.0	1.5		0.002		0.05				
		- as in first argillite but more carbon rich	108573	87.0	88.5	1.5		0.002		0.05				
		- graphite along shear planes at 55° to core axis	108574	88.5	90.0	1.5		0.002		0.04				
		- bedding at 55° to core axis cross cutting shear planes	108575	90.0	91.5	1.5		0.002		0.03				
		- 3 - 5% pyrite in coarser, pyrite rich layers												
		- occasional calcite veins at 45° to core axis up to 0.1 cm wide												
		- crystallites in darker layers												
		90.3												
		- fossil fragment												
		85.8 - 86.1												
		- 20% andesite fragments												
		- rubbly												
91.8	122.8	ANDESITE FLOWS	109576	91.5	93.0	1.5		<0.002		<0.02				
		- light to medium green	109577	93.0	94.5	1.5		<0.002		<0.02				
		- aphanitic to fine grained phaneritic	109578	94.5	96.0	1.5		<0.002		<0.02				

From	To	Description	Sample No.	Sample		Length	Au ppb	Au oz/t	Ag ppm	ASSAYS				
				From	To					Ag oz/t	Pb ppm	Cu ppm	Zn ppm	As ppm
		- locally porphyritic	109579	96.0	97.5	1.5		<0.002		0.03				
		- locally amygdular	109580	97.5	98.0	1.5		<0.002		0.02				
		- occasional crackle breccias with grey calcite, trace chlorite, trace pyrite as matrix	109581	98.0	100.5	1.5		<0.002		<0.02				
			109582	100.5	102.0	1.5		<0.002		<0.02				
			109583	102.0	103.5	1.5		<0.002		<0.02				
		91.8 - 99.1	109584	103.5	105.0	1.5		<0.002		<0.02				
		- andesite breccia	109585	105.0	106.5	1.5		<0.002		<0.02				
		- aphanitic to above andesite type clasts in a black argillite	109586	106.5	108.0	1.5		<0.002		<0.02				
			109587	108.0	109.5	1.5		<0.002		<0.02				
		- matrix	109588	109.5	111.0	1.5		<0.002		<0.02				
		- occasional argillite zones	109589	111.0	112.5	1.5		<0.002		<0.02				
		95.1 to 97.8, 99.1 - 99.4	109590	112.5	114.0	1.5		<0.002		<0.02				
		- poor recovery (40% in 95.1 to 97.8 zone) no distinct bedding planes	109591	114.0	115.5	1.5		<0.002		<0.02				
			109592	115.5	117.0	1.5		<0.002		<0.02				
		- calcite veining and shearing at 80° to core axis	109593	117.0	118.5	1.5		<0.002		<0.02				
			109594	118.5	120.0	1.5		<0.002		<0.02				
		- rubble zones:	109595	120.0	121.5	1.5		<0.002		<0.02				
		112.0 - 115.8 oxidized fractures	109596	121.5	123.0	1.5		<0.002		<0.02				
		121.8 - 121.1 rubbly oxidized fracture zone												
		122.5 - 122.8 as above												
122.8	230.7	RHYOLITE UNIT	109597	123.0	124.5	1.5		<0.002		<0.02				
		- siliceous rhyolite breccia ash tuff units	109598	124.5	126.0	1.5		<0.002		<0.02				
			109599	126.0	127.5	1.5		<0.002		<0.02				
		- contact with andesite intensely fractured rubbly and oxidized 122.8 to 123.6 (poor recovery 40%)	109600	127.5	129.0	1.5		<0.002		<0.02				
			109601	129.0	130.5	1.5		<0.002		<0.02				
			109602	130.5	132.0	1.5		<0.002		<0.02				
		- no argillite	109603	132.0	133.5	1.5		<0.002		<0.02				
		- pale grey rhyolite clasts 0.5 cm to 8 cm across in medium grey fine grained matrix	109604	133.5	135.0	1.5		<0.002		<0.02				
			109605	135.0	136.5	1.5		<0.002		<0.02				
			109606	136.5	138.0	1.5		<0.002		<0.02				
		- intense micro-fracturing, fractured 1 per cm	109607	138.0	139.5	1.5		<0.002		<0.02				
			109608	139.5	141.0	1.5		<0.002		<0.02				
		- occasional oxidized fractures	109609	141.0	142.5	1.5		<0.002		<0.02				

From	To	Description	Sample No.	From	To	Length	Au ppb	Au oz/t	Ag ppm	ASSAYS				
										Ag oz/t	Pb ppm	Cu ppm	Zn ppm	As ppm
		35.3 - 49.2, 57.5 - 59.2, 59.8 - 65.5	109685	55.5	57.0	1.5		<0.002		0.02				
		- banded silstone argillite	109686	57.0	58.5	1.5		<0.002		<0.02				
		- layers at 20° to core axis in 35.3 to 49.2	109687	58.5	60.0	1.5		0.002		0.03				
		- average of 1 cm width	109688	61.5	63.0	1.5		0.002		0.03				
		- layers denoted by lighter coarser grained	109689	61.5	63.0	1.5		<0.002		0.02				
		layers & darker grey black carbonaceous	109690	63.0	64.5	1.5		<0.002		<0.02				
		finer grained layers with occasional	109691	64.5	66.0	1.5		<0.002		<0.02				
		crystallites	109692	66.0	67.5	1.5		<0.002		<0.02				
		- 1 - 3% pyrite as blebs, disseminations &	109693	75.0	78.5	1.5		<0.002		<0.02				
		fracture filling associated with calcite	109694	78.1	79.5	1.5		<0.002		0.02				
		- trace sphalerite in calcite filled fracture												
		at 41.0 & 59.9												
		- fractures 20° to 50° to core axis												
		- no apparent bedding in 57.5 to 59.2 as in												
		59.8 to 63.5												
		- calcite fracture filling up to 1 cm wide												
		- frequency of 1 per 10 cm at 30° to 50°												
		to core axis												
		59.2 - 50.9 Cave												
		- 50.9 - 57.5, 59.2 - 59.8, 63.5 - 64.7												
		andesite flow breccia												
		- fine grained aphanitic with matrix of												
		grey calcite and pyrite												
		- argillite matrix in outer 10 cm of breccia												
		64.7 - 79.3												
		- andesite flows												
		- fine grained phaneritic with aphanitic												
		chill margins, no quench rinds												
		- 1 - 3% pyrite as part of matrix												
		(including grey calcite, trace chlorite)												
		in crackle breccia												
79.3	84.7	GRAPHITIC ARGILLITE	109695	79.5	81.0	1.5		0.004		0.06				
		- 1 - 3% pyrite	109696	81.0	82.5	1.5		0.002		0.02				
		- layering at 40° to core axis	109697	82.5	84.0	1.5		0.002		0.02				

From	To	Description	Sample No.	Sample		Length	Au ppb	Au oz/t	Ag ppm	ASSAYS					
				From	To					Ag oz/t	Pb ppm	Cu ppm	Zn ppm	As ppm	Sb ppm
		- calcite fracture filling at 15° to core axis up to 0.3 mm wide	109698	94.0	85.5	1.5		<0.002		<0.02					
		- frequency of 1 per 0.5 m													
84.7	89.3	ANDESITE BRECCIA	109699	85.5	87.0	1.5		<0.002		<0.02					
		- andesite fragments up to 40cm wide in an argillite matrix	109700	87.0	88.5	1.5		<0.002		<0.02					
		- 40% argillite	109701	88.5	90.0	1.5		<0.002		0.03					
		- 85.2 - 85.7, 86.9 - 87.4 argillite, no andesite fragments													
		- 3 -5% pyrite as blebs and disseminated in the matrix & rimming 5% of the fragments													
		- fragments angular to subangular													
		89.1 - 89.3													
		- sheared argillite with graphite along shear planes ranging from 30 to 80° to core axis contact with rhyolite at 80° to core axis													
89.3		RHYOLITE UNIT	109702	90.0	91.5	1.5		<0.002		<0.02					
		89.3 - 96.0	109703	91.5	93.0	1.5		<0.002		<0.02					
		- pale grey green dust tuff	109704	93.0	94.5	1.5		<0.002		<0.02					
		- no discernable fragment size	109705	94.5	96.0	1.5		<0.002		<0.02					
		- layering at 45° to core axis (foliations also)	109706	96.0	97.5	1.5		<0.002		<0.02					
			109707	97.5	99.0	1.5		<0.002		<0.02					
		- 1 - 3% pyrite as disseminations & blebs	109708	99.0	100.5	1.5		<0.002		<0.02					
		- occasional oxidized fractures	109709	100.5	102.0	1.5		<0.002		<0.02					
		90.3 - 90.6	109710	102.0	103.5	1.5		<0.002		<0.02					
		- intensely silicified, massive dust tuff	109711	103.5	105.0	1.5		<0.002		<0.02					
		91.6 - 91.7	109712	105.0	106.5	1.5		<0.002		<0.02					
		- mud seam at 50° to core axis	109713	106.5	108.0	1.5		<0.002		<0.02					
		93.9 - 94.0	109714	108.0	109.5	1.5		<0.002		<0.02					
		- mud seam at 70° to core axis	109715	109.5	111.0	1.5		<0.002		<0.02					
			109716	111.0	112.5	1.5		<0.002		<0.02					

From	To	Description	Sample		Length	ASSAYS									
			No.	From To		Au ppb	Au oz/t	Ag ppm	Ag oz/t	Pb ppm	Cu ppm	Zn ppm	As ppm	Sb ppm	
94.5	95.1		109717	112.5 114.0	1.5		<0.002		<0.02						
		- 60% quartz - 10% calcite flooding	109718	114.0 115.5	1.5		<0.002		<0.02						
96.0	125.3	Siliceous	109719	115.5 117.0	1.5		<0.002		<0.02						
		- light grey ash tuff units	109720	117.0 118.5	1.5		<0.002		<0.02						
		- fragments 2 - 6 cm fining upwards	109721	118.5 120.0	1.5		<0.002		<0.02						
		- intensely fractured, two major fracture sets at 10° to core axis & at 30° to core axis	109722	120.0 121.5	1.5		<0.002		<0.02						
			109723	121.5 123.0	1.5		<0.002		<0.02						
			109724	123.0 124.5	1.5		<0.002		<0.02						
		- frequency of 1 per cm	109725	124.5 126.0	1.5		<0.002		<0.02						
		- 1 - 3% pyrite as above but also as occasional fracture filling	109726	126.0 127.5	1.5		<0.002		<0.02						
			109727	127.5 129.0	1.5		<0.002		<0.02						
		- fractures are mostly oxidized & leached to 1 cm away from fracture from 96.0 - 111.1, 117.3 - 117.6	109728	129.0 130.5	1.5		<0.002		<0.02						
			109729	130.5 135.0	1.5		<0.002		<0.02						
		- stylolitic fractures from 109.5 - 125.8 filled with pyrite, calcite & chlorite	109730	133.5 135.0	1.5		<0.002		<0.02						
			109731	133.5 135.0	1.5		<0.002		<0.02						
		- quartz fracture filling (flooding) from 111.2 - 114.9 (as well as minor calcite)	109732	135.0 136.5	1.5		<0.002		<0.02						
			109733	136.5 138.0	1.5		<0.002		<0.02						
			109734	138.0 139.5	1.5		<0.002		<0.02						
117.1	117.3		109735	139.5 141.0	1.5		<0.002		<0.02						
		- mud seam, rubbly	109736	141.0 142.5	1.5		<0.002		<0.02						
125.8	142.3		109737	142.5 144.0	1.5		<0.002		<0.02						
		- rhyolite breccia (auto breccia)	109738	144.0 145.5	1.5		<0.002		<0.02						
		- pale grey	109739	145.5 147.0	1.5		<0.002		<0.02						
		- 1 - 3% pyrite as blebs and occasional fracture filling	109740	147.0 148.5	1.5		<0.002		<0.02						
			109741	148.5 150.0	1.5		<0.002		<0.02						
		- foliations (weak) at 40° to core axis	109742	150.0 151.5	1.5		<0.002		<0.02						
		- occasional oxidized fractures	109743	151.5 153.0	1.5		<0.002		<0.02						
		- low fracture density 1 per 30 cm	109744	153.0 154.5	1.5		<0.002		<0.02						
142.3	142.8	Andesite Dike	109745	154.5 156.0	1.5		<0.002		<0.02						
		- pale green	109746	156.0 157.5	1.5		<0.002		<0.02						
		- amygdules filled with chlorite & calcite	109747	157.5 159.0	1.5		<0.002		<0.02						
		- sericitic? alteration from 141.5 - 142.3 & from 142.8 - 143.7 (halo around dyke)	109748	159.0 160.5	1.5		<0.002		<0.02						
			109749	160.5 162.0	1.5		<0.002		<0.02						
		(pale yellow, soft, clayish mineral disseminated flecks)	109750	162.0 163.5	1.5		<0.002		<0.02						
			109751	163.5 165.0	1.5		<0.002		<0.02						

From	To	Description	Sample		Length	Au ppb	Au oz/t	ASSAYS						
			No.	From To				Ag ppm	Ag oz/t	Pb ppm	Cu ppm	Zn ppm	As ppm	Sb ppm
142.8 - 224.6		Rhyolite Ash Tuff	109752	165.0 166.5	1.5		<0.002	<0.02						
		- clast size from 2 - 6 cm	109753	166.5 168.0	1.5		<0.002	<0.02						
		- angular to subangular	109754	168.0 169.5	1.5		<0.002	<0.02						
		- increase in fracture density to 1 per cm	109756	171.0 172.5	1.5		<0.002	<0.02						
		- no preferred orientation	109757	172.5 174.0	1.5		<0.002	<0.02						
		- pale grey to medium grey	109758	174.0 175.5	1.5		<0.002	<0.02						
		- siliceous	109759	175.5 177.0	1.5		<0.002	<0.02						
		- 30% fragments are flow banded	109760	177.0 178.5	1.5		<0.002	<0.02						
		- 1% pyrite as blebs and occasional fracture filling with calcite	109761	178.5 180.0	1.5		<0.002	<0.02						
			109762	180.0 181.5	1.5		<0.002	<0.02						
159.8 - 162.4			109763	181.5 183.0	1.5		<0.002	<0.02						
		- sericite alteration	109764	183.0 184.5	1.5		<0.002	<0.02						
		- pale yellow flecks as in 141.5 to 142.3	109765	184.5 186.0	1.5		<0.002	<0.02						
		- occasional zones of fine ash tuff up to 30 cm wide	109766	186.0 187.5	1.5		<0.002	<0.02						
			109767	187.5 189.0	1.5		<0.002	<0.02						
175.1 - 218.6			109768	189.0 190.5	1.5		<0.002	<0.02						
		- 5% sericite (yellow clayey flecks) alteration of clasts	109769	190.5 192.0	1.5		<0.002	<0.02						
			109770	192.0 193.5	1.5		<0.002	<0.02						
		- moderately foliated with foliations at 40° to core axis	109771	193.5 195.0	1.5		<0.002	<0.02						
			109772	195.0 196.5	1.5		<0.002	<0.02						
		- pyrite along with calcite as blebs and along dendritic (stylolite-like) fractures throughout	109773	196.5 198.0	1.5		<0.002	<0.02						
			109774	198.0 199.5	1.5		<0.002	<0.02						
			109775	199.5 201.0	1.5		<0.002	<0.02						
168.8 - 224.6			109776	201.0 202.5	1.5		<0.002	<0.02						
		- calcite filled fractures (tension gashes) 1 per 2 cm	109777	202.5 204.0	1.5		<0.002	<0.02						
			109778	204.0 205.5	1.5		<0.002	<0.02						
170.6 - 170.7, 201.5 - 201.6		Andesite	109779	205.5 207.0	1.5		<0.002	<0.02						
		Dykes	109780	207.0 208.5	1.5		<0.002	<0.02						
		204.6 - foliations at 30° to core axis	109781	208.5 210.0	1.5		<0.002	<0.02						
		206.0 - foliations at 25° to core axis	109782	210.0 211.5	1.5		<0.002	<0.02						
		198.8 - 199.7 - 5% clasts in dust tuff, clasts have pale yellow alteration rims	109783	211.5 213.0	1.5		<0.002	<0.02						
			109784	213.0 214.5	1.5		<0.002	<0.02						
		197.3 - 201.5, 205.1 - 207.7, 211.9 -	109785	214.5 216.0	1.5		<0.002	<0.02						
		224.6 - emerald green clasts in grey ash matrix	109786	216.0 217.5	1.5		<0.002	<0.02						
			109787	217.5 219.0	1.5		<0.002	<0.02						

CALPINE RESOURCES INCORPORATED - ESKAY CREEK PROPERTY, B.C.

DRILL HOLE LOG Hole Number CA89-38

Azimuth:	150	Elevation:	99.12 metres
Dip:	-45		325.2 feet
Core Size:	NQ		
		Length:	154.4 metres
			506.6 feet
Started:	March 7, 1989	Line:	2+41S 1+14W
Completed:	March 10, 1989		
		Location:	969.45 Northing
			708.70 Easting

Drilling Company: Falcon Drilling
Logged by: G.F. McArthur
Assayed by: Bondar-Clegg

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Certificate
of Analysis



Bondar-Clegg & Company Ltd.
130 Pemberton Ave.
North Vancouver, B.C.
V 1R5
(604) 985-0681 Telex 04-352667

REPORT: V69-01018.4

PROJECT: UNUK

PAGE 1

SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT	SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT
				D2 109386		0.003	0.05
				D2 109387		0.003	0.06
				D2 109388		0.003	0.14
				D2 109389		0.008	0.10
D2 109350	<0.002	<0.02		D2 109390		0.010	1.75
D2 109351	<0.002	<0.02		D2 109391		0.006	0.11
D2 109352	<0.002	<0.02		D2 109392		0.003	0.06
D2 109353	<0.002	<0.02		D2 109393		0.006	0.13
D2 109354	<0.002	<0.02		D2 109394		0.002	0.19
D2 109355	<0.002	<0.02		D2 109395		0.003	0.11
D2 109356	<0.002	<0.02		D2 109396		0.004	0.06
D2 109357	<0.002	<0.02		D2 109397		0.002	0.02
D2 109358	<0.002	<0.02		D2 109398		0.002	0.06
D2 109359	<0.002	0.02		D2 109399		<0.002	0.23
D2 109360	<0.002	0.03		D2 109400		0.002	0.16
D2 109361	<0.002	<0.02		D2 109401		<0.002	0.23
D2 109362	<0.002	<0.02		D2 109402		0.002	0.06
D2 109363	0.002	<0.02		D2 109403		0.002	0.17
D2 109364	0.002	<0.02		D2 109404		0.005	0.69
D2 109365	<0.002	<0.02		D2 109405		0.015	0.76
D2 109366	0.004	0.08		D2 109406		0.002	0.08
D2 109367	<0.002	<0.02		D2 109407		0.003	0.09
D2 109368	<0.002	<0.02		D2 109408		0.008	0.73
D2 109369	<0.002	<0.02		D2 109409		0.003	0.03
D2 109370	<0.002	<0.02		D2 109410		0.002	0.02
D2 109371	<0.002	<0.02		D2 109411		<0.002	0.02
D2 109372	<0.002	<0.02		D2 109412		0.002	0.03
D2 109373	0.002	<0.02		D2 109413		<0.002	0.03
D2 109374	<0.002	<0.02		D2 109414		<0.002	0.03
D2 109375	<0.002	<0.02		D2 109415		<0.002	0.04
D2 109376	<0.002	<0.02		D2 109416		<0.002	0.03
D2 109377	<0.002	<0.02		D2 109417		<0.002	0.02
D2 109378	<0.002	<0.02		D2 109418		<0.002	0.02
D2 109379	<0.002	<0.02		D2 109419		0.002	0.03
D2 109380	<0.002	<0.02		D2 109420		0.002	0.03
D2 109381	<0.002	<0.02		D2 109421		0.002	0.03
D2 109382	0.002	0.04		D2 109422		0.002	0.02
D2 109383	<0.002	0.02		D2 109423		0.003	0.03
D2 109384	<0.002	0.03		D2 109424		0.002	0.04
D2 109385	0.004	0.05		D2 109425		<0.002	0.02

CA-30

CA-38

Bondar-Clegg & Company Ltd.
10 Pemberton Ave.
Vancouver, B.C.
Tel: 5-0681 Telex (4-352887)



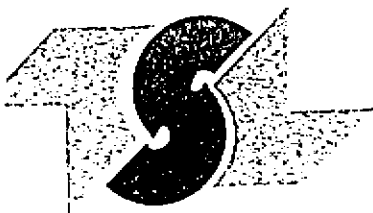
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of Analysis

REPORT: V89-01018.4

PROJECT: UNUK

PAGE 2

SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT	SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT
D2 109426		<0.002	0.02				
D2 109427		<0.002	0.03				
D2 109428		0.002	0.02				
D2 109429	CA-38	0.002	0.02				
D2 109430		0.002	0.02				
D2 109431		<0.002	0.02				
D2 109432		<0.002	0.02				



ENTERED APR 24 1989
TSL LABORATORIES
DIV. BURGNER TECHNICAL ENTERPRISES LIMITED

2 - 302 - 45th STREET, EAST
SASKATOON, SASKATCHEWAN
S7K 6A4

☎ (306) 931-1033 FAX: (306) 242-4717

CERTIFICATE OF ANALYSIS

SAMPLE(S) FROM Prime Exploration Ltd.
10th Floor-Box 10, 808 West Hastings St.
Vancouver, B.C.
.V6C 2X6

REPORT No.
S6457

SAMPLE(S) OF Pulps

INVOICE #: 11157
P.O.: Bondar-Clegg

Recheck

Keewatin Engineering

	Au ozt	Ag ozt
.109360	<.001	<.05
.09375	<.001	<.05
109390	.013	1.56
.109405	.019	.69
.09420	.001	<.05

CA-38

CALPINE RESOURCES INCORPORATED - ESKAY CREEK PROPERTY, B.C.

DRILL HOLE LOG Hole Number CA89-39

Azimuth:	150	Elevation:	99.12	metres
Dip:	-77		325.2	feet
Core Size:	NQ			
		Length:	203.3	metres
			667.0	feet
Started:	March 10, 1989	Line:	2+41S	1+14W
Completed:	March 14, 1989	Location:	969.45	Northing
			708.70	Easting

Drilling Company: Falcon Drilling
Logged by: Christine Swanson
Assayed by: Bondar-Clegg

Bondar-Clegg & Company Ltd.
 130 Pemberton Ave.
 New Vancouver, B.C.
 V7Y 2R5
 (604) 985-0681 Telex 04-352667



REPORT: U89-01018.4

PROJECT: UNUK PAGE 2

SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT	SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT
				D2 109484		<0.002	<0.02
				D2 109485		<0.002	<0.02
				D2 109486		<0.002	<0.02
				D2 109487		<0.002	<0.02
				D2 109488		<0.002	<0.02
				D2 109489		<0.002	<0.02
				D2 109490		<0.002	<0.02
D2 109451		<0.002	<0.02	D2 109491		<0.002	<0.02
D2 109452		<0.002	<0.02	D2 109492		<0.002	<0.02
D2 109453		<0.002	<0.02	D2 109493		0.002	<0.02
D2 109454		<0.002	<0.02	D2 109494		<0.002	<0.02
D2 109455		<0.002	<0.02	D2 109495		<0.002	<0.02
D2 109456		<0.002	<0.02	D2 109496		<0.002	<0.02
D2 109457		<0.002	<0.02	D2 109497	CA-39	<0.002	<0.02
D2 109458		<0.002	0.05	D2 109498		<0.002	<0.02
D2 109459		<0.002	0.02	D2 109499		<0.002	<0.02
D2 109460		<0.002	<0.02	D2 109500		<0.002	<0.02
D2 109461		<0.002	<0.02	D2 109501		<0.002	<0.02
D2 109462		<0.002	<0.02	D2 109502		0.003	<0.02
D2 109463	CA-39	<0.002	<0.02	D2 109503		<0.002	<0.02
D2 109464		<0.002	<0.02	D2 109504		0.002	<0.02
D2 109465		<0.002	<0.02	D2 109505		0.004	<0.02
D2 109466		<0.002	<0.02	D2 109506		<0.002	<0.02
D2 109467		<0.002	<0.02	D2 109507		<0.002	<0.02
D2 109468		<0.002	<0.02	D2 109508		<0.002	<0.02
D2 109469		<0.002	0.02	D2 109509		0.002	0.02
D2 109470		<0.002	<0.02	D2 109510		<0.002	<0.02
D2 109471		<0.002	<0.02	D2 109511		<0.002	<0.02
D2 109472		<0.002	<0.02	D2 109512		<0.002	<0.02
D2 109473		<0.002	<0.02	D2 109513		<0.002	<0.02
D2 109474		<0.002	<0.02	D2 109514		<0.002	<0.02
D2 109475		<0.002	<0.02	D2 109515		<0.002	<0.02
D2 109476		<0.002	<0.02				
D2 109477		0.003	<0.02				
D2 109478		<0.002	<0.02				
D2 109479		<0.002	<0.02				
D2 109480		<0.002	<0.02				
D2 109481		<0.002	<0.02				
D2 109482		<0.002	0.02				
D2 109483		<0.002	<0.02				

Bondar-Clegg & Company Ltd.
 130 Pemberton Ave.
 North Vancouver, B.C.
 V7J 1R5
 (604) 985-0681 Telex 04-352667



Certificate
 of Analysis

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 Apr 7/89

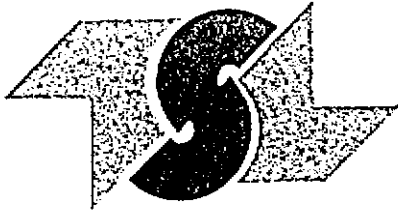
REPORT: V89-D1062.4

PROJECT: UNUK

PAGE 1

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D2 109517		<0.002	<0.02				
D2 109518		<0.002	<0.02				
D2 109519		<0.002	<0.02				
D2 109520		<0.002	<0.02				
D2 109521		<0.002	<0.02				
D2 109522		<0.002	<0.02				
D2 109523		<0.002	<0.02				
D2 109524		<0.002	<0.02				
D2 109525		<0.002	<0.02				
D2 109526		<0.002	0.02				
D2 109527		<0.002	<0.02				
D2 109528		<0.002	<0.02				
D2 109529		<0.002	<0.02				
D2 109530		<0.002	<0.02				
D2 109531	CA-39 ↓	<0.002	<0.02				
D2 109532		<0.002	<0.02				
D2 109533		<0.002	<0.02				
D2 109534		<0.002	<0.02				
D2 109535		<0.002	<0.02				
D2 109536		<0.002	<0.02				
D2 109537		<0.002	<0.02				
D2 109538		<0.002	<0.02				
D2 109539		<0.002	<0.02				
D2 109540		<0.002	<0.02				
D2 109541		<0.002	<0.02				
D2 109542		<0.002	<0.02				
D2 109543		<0.002	0.02				
D2 109544		<0.002	<0.02				
D2 109545		<0.002	<0.02				
D2 109546	↓	<0.002	<0.02				
D2 109547		<0.002	0.03				
D2 109548		<0.002	<0.02				

ENTERED APR 24 1989



TSL LABORATORIES

DIV. BURGNER TECHNICAL ENTERPRISES LIMITED

2 - 302 - 45th STREET, EAST
SASKATOON, SASKATCHEWAN
S7K 6A4

☎ (306) 931-1033 FAX: (306) 242-4717

CERTIFICATE OF ANALYSIS

SAMPLE(S) FROM Prime Exploration Ltd.
10th Floor-Box 10, 808 West Hastings St.
Vancouver, B.C.
V6C 2X6

REPORT No.
S6457

SAMPLE(S) OF Pulps

INVOICE #: 11157
P.O.: Bondar-Clegg

Recheck

Keewatin Engineering

Au
ozt

Ag
ozt

	Au ozt	Ag ozt
109453	<.001	<.05
109468	.007/<.001	<.05
109483	.001	<.05
109498	<.001	<.05
109513	<.001	<.05

CA-39

COPIES TO: C. Idziszek, J. Foster
INVOICE TO: Keewatin Engineering-Vancouver

Mar 23/89

SIGNED

Bernie Owen



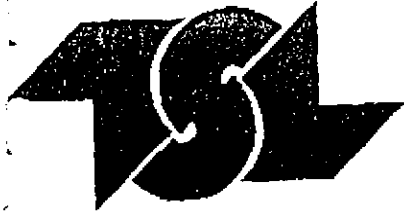
ENTERED APR 24 1989

TSL LABORATORIES

DIV. BURGNER TECHNICAL ENTERPRISES LIMITED

2-302 - 48th STREET, EAST
SASKATOON, SASKATCHEWAN
S7K 6A4

☎ (306) 931-1033 FAX: (306) 242-4717



CERTIFICATE OF ANALYSIS

SAMPLE(S) FROM Prime Exploration Ltd.
10th Floor-Box 10, 808 West Hastings St.
Vancouver, B.C.
V6C 2X6

REPORT No.
S6542

INVOICE #: 11253
P.O.:

SAMPLE(S) OF Pulps

Recheck

Bondar-Clegg

	Au ozt		Ag ozt
V89-01062.4-109530	<.001		<.05
V89-01062.4-109545	<.001/.001	CA-39	<.05

CALPINE RESOURCES INCORPORATED - ESKAY CREEK PROPERTY, B.C.

DRILL HOLE LOG Hole Number CA89-40

Azimuth:	330	Elevation:	99.12	metres
Dip:	-78		325.2	feet
Core Size:	NQ			
		Length:	230.7	metres
			757.0	feet
Started:	March 14, 1989	Line:	2+41S	1+14W
Completed:	March 17, 1989			
		Location:	969.45	Northing
			708.70	Easting

Drilling Company: Falcon Drilling
Logged by: Christine Swanson
Assayed by: Bondar-Clegg

Bondar-Clegg & Company Ltd.
 10 Pemberton Ave.
 North Vancouver, B.C.
 P.O. Box 100
 (604) 955-0851 Telex 04-352667



Certificate
 of Analysis

ENTERED APR 07 1980

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REPORT: V87-01062.4

PROJECT: UNUK

PAGE 1

SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT	SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT
				D2 109556		<0.002	<0.02
				D2 109557		<0.002	<0.02
				D2 109558		<0.002	0.02
				D2 109559		<0.002	0.03
				D2 109560		<0.002	<0.02
				D2 109561		<0.002	<0.02
				D2 109562		<0.002	0.02
				D2 109563		<0.002	0.03
				D2 109564		<0.002	0.03
				D2 109565		<0.002	0.02
				D2 109566		<0.002	0.02
				D2 109567		<0.002	<0.02
				D2 109568		<0.002	0.02
				D2 109569		<0.002	<0.02
				D2 109570		0.003	<0.02
				D2 109571		<0.002	<0.02
				D2 109572		0.002	0.05
				D2 109573		0.002	0.05
				D2 109574		0.002	0.04
				D2 109575		0.002	0.03
				D2 109576		<0.002	<0.02
				D2 109577		<0.002	<0.02
				D2 109578		<0.002	<0.02
				D2 109579		<0.002	0.03
				D2 109580		<0.002	0.02
				D2 109581		<0.002	<0.02
				D2 109582		<0.002	<0.02
				D2 109583		<0.002	<0.02
				D2 109584		<0.002	<0.02
				D2 109585		<0.002	<0.02
				D2 109586		<0.002	<0.02
				D2 109587		<0.002	<0.02
				D2 109588		<0.002	<0.02
				D2 109589		<0.002	<0.02
				D2 109590		<0.002	<0.02
D2 109549		<0.002	<0.02				
D2 109550		<0.002	<0.02				
D2 109551		<0.002	<0.02	D2 109591		<0.002	<0.02
D2 109552	CA-40	<0.002	0.02	D2 109592		<0.002	<0.02
D2 109553		<0.002	0.04	D2 109593		<0.002	<0.02
D2 109554		<0.002	0.02	D2 109594		<0.002	<0.02
D2 109555		<0.002	<0.02	D2 109595		<0.002	<0.02

CA-40



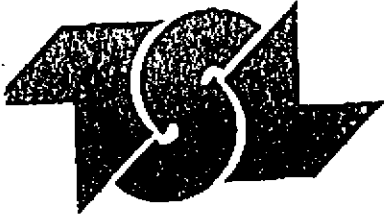
ENTERED APR 07 1989

REPORT: V59-01062.4

PROJECT: JNUK

PAGE 2

SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT	SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT
D2 109596		<0.002	<0.02	D2 109636		<0.002	<0.02
D2 109597		<0.002	<0.02	D2 109637		<0.002	<0.02
D2 109598		<0.002	<0.02	D2 109638		<0.002	<0.02
D2 109599		<0.002	<0.02	D2 109639		<0.002	<0.02
D2 109600		<0.002	<0.02	D2 109640		<0.002	<0.02
D2 109601		<0.002	<0.02	D2 109641		<0.002	<0.02
D2 109602		<0.002	<0.02	D2 109642		<0.002	<0.02
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D2 109606		<0.002	<0.02	D2 109646		<0.002	<0.02
D2 109607		<0.002	<0.02	D2 109647		<0.002	<0.02
D2 109608		<0.002	<0.02	D2 109648		<0.002	<0.02
D2 109609		<0.002	<0.02	D2 109649		<0.002	<0.02
D2 109610		<0.002	<0.02	D2 109650		<0.002	<0.02
D2 109611		<0.002	<0.02	D2 109651	CA-40	<0.002	<0.02
D2 109612		<0.002	<0.02	D2 109652		<0.002	<0.02
D2 109613		<0.002	<0.02	D2 109653		<0.002	<0.02
D2 109614	CA-40	<0.002	<0.02	D2 109654		<0.002	<0.02
D2 109615		<0.002	<0.02	D2 109655		<0.002	<0.02
D2 109616		0.002	<0.02	D2 109656		<0.002	<0.02
D2 109617		<0.002	<0.02	D2 109657		<0.002	<0.02
D2 109618		<0.002	<0.02	D2 109658		<0.002	<0.02
D2 109619		<0.002	0.03	D2 109659		<0.002	<0.02
D2 109620		<0.002	<0.02	D2 109660		<0.002	0.02
D2 109621		<0.002	<0.02	D2 109661		<0.002	<0.02
D2 109622		<0.002	<0.02	D2 109662		<0.002	<0.02
D2 109623		<0.002	<0.02	D2 109663		<0.002	<0.02
D2 109624		<0.002	<0.02	D2 109664		<0.002	<0.02
D2 109625		<0.002	<0.02	D2 109665		<0.002	<0.02
D2 109626		<0.002	<0.02	D2 109666		<0.002	<0.02
D2 109627		<0.002	<0.02	D2 109667		<0.002	<0.02
D2 109628		<0.002	<0.02	D2 109668		<0.002	<0.02
D2 109629		<0.002	<0.02				
D2 109630		<0.002	<0.02				
D2 109631		<0.002	<0.02				
D2 109632		<0.002	<0.02				
D2 109633		<0.002	<0.02				
D2 109634		<0.002	<0.02				
D2 109635		<0.002	<0.02				



ENTERED APR 21 1989

TSL LABORATORIES

DIV. BURGNER TECHNICAL ENTERPRISES LIMITED

2 - 302 - 46th STREET, EAST
SASKATOON, SASKATCHEWAN
S7K 6A4

☎ (306) 931-1033 FAX: (306) 242-4717

CERTIFICATE OF ANALYSIS

SAMPLE(S) FROM Prime Exploration Ltd.
10th Floor-Box 10, 808 West Hastings St.
Vancouver, B.C.
V6C 2X6

REPORT No.
S6542

INVOICE #: 11253
P.O.:

SAMPLE(S) OF Pulps

Recheck

Bondar-Clegg

	Au ozt		Ag ozt
V89-01062.4-109560	<.001		<.05
V89-01062.4-109575	<.001		<.05
V89-01062.4-109590	.004		<.05
V89-01062.4-109605	<.001		<.05
V89-01062.4-109620	<.001	CA-40	<.05
V89-01062.4-109635	<.001		<.05
V89-01062.4-109650	<.001		<.05
V89-01062.4-109665	<.001		<.05

CALPINE RESOURCES INCORPORATED - ESKAY CREEK PROPERTY, B.C.

DRILL HOLE LOG Hole Number CA89-41

Azimuth:	330	Elevation:	99.12 metres
Dip:	-60		325.2 feet
Core Size:	NQ	Length:	224.6 metres
			737.0 feet
Started:	March 18, 1989	Line:	2+41S 1+14W
Completed:	March 22, 1989	Location:	969.45 Northing
			708.70 Easting
Drilling Company:	Falcon Drilling		
Logged by:	Christine Swanson		
Assayed by:	Bondar-Clegg		

Bondar-Clegg & Company Ltd.
30 Pemberton Ave.
North Vancouver, B.C.
V7P 2R5
(604) 985-0681 Telex 04-352667



Certificate of Analysis

ENTERED

REPORT: V39-01062.4

PROJECT: UNUK

PAGE 2

SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT	SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT
				D2 109669		<0.002	<0.02
				D2 109670		<0.002	<0.02
				D2 109671		0.004	<0.02
				D2 109672	CA-41	<0.002	0.03
				D2 109673		<0.002	<0.02
				D2 109674		<0.002	0.02
				D2 109675		<0.002	0.02



Registered Assayer, Province of British Columbia



Apr. 7/89

REPORT: V89-01062.4

PROJECT: UNUK

PAGE 3

SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT	SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT
D2 109676		<0.002	0.03	D2 109716		<0.002	<0.02
D2 109677		0.002	0.05	D2 109717		<0.002	<0.02
D2 109678		<0.002	0.04	D2 109718		<0.002	<0.02
D2 109679		0.002	0.03	D2 109719		<0.002	<0.02
D2 109680		0.002	0.04	D2 109720		<0.002	<0.02
D2 109681		0.002	0.05	D2 109721		<0.002	<0.02
D2 109682		<0.002	0.02	D2 109722		<0.002	<0.02
D2 109683		<0.002	<0.02	D2 109723		<0.002	<0.02
D2 109684		<0.002	0.02	D2 109724		<0.002	<0.02
D2 109685		<0.002	0.02	D2 109725		<0.002	<0.02
D2 109686		<0.002	<0.02	D2 109726		<0.002	<0.02
D2 109687		0.002	0.03	D2 109727		<0.002	<0.02
D2 109688		0.002	0.03	D2 109728		<0.002	<0.02
D2 109689		<0.002	0.02	D2 109729		<0.002	<0.02
D2 109690		<0.002	<0.02	D2 109730		<0.002	<0.02
D2 109691		<0.002	<0.02	D2 109731		<0.002	<0.02
D2 109692		<0.002	<0.02	D2 109732		<0.002	<0.02
D2 109693	CA-41	<0.002	<0.02	D2 109733	CA-41	<0.002	<0.02
D2 109694		<0.002	0.02	D2 109734		<0.002	<0.02
D2 109695		0.004	0.06	D2 109735		<0.002	<0.02
D2 109696		0.002	0.02	D2 109736		<0.002	<0.02
D2 109697		0.002	0.02	D2 109737		<0.002	<0.02
D2 109698		<0.002	<0.02	D2 109738		<0.002	<0.02
D2 109699		<0.002	<0.02	D2 109739		<0.002	<0.02
D2 109700		<0.002	<0.02	D2 109740		<0.002	<0.02
D2 109701		<0.002	0.03	D2 109741		<0.002	<0.02
D2 109702		<0.002	<0.02	D2 109742		<0.002	<0.02
D2 109703		<0.002	<0.02	D2 109743		<0.002	<0.02
D2 109704		<0.002	<0.02	D2 109744		<0.002	<0.02
D2 109705		<0.002	<0.02	D2 109745		<0.002	<0.02
D2 109706		<0.002	<0.02	D2 109746		<0.002	<0.02
D2 109707		<0.002	<0.02	D2 109747		<0.002	<0.02
D2 109708		<0.002	<0.02	D2 109748		<0.002	<0.02
D2 109709		<0.002	<0.02	D2 109749		<0.002	<0.02
D2 109710		<0.002	<0.02	D2 109750		<0.002	<0.02
D2 109711		<0.002	<0.02	D2 109751		<0.002	<0.02
D2 109712		<0.002	<0.02	D2 109752		<0.002	<0.02
D2 109713		<0.002	<0.02	D2 109753		<0.002	<0.02
D2 109714		<0.002	<0.02	D2 109754		<0.002	<0.02
D2 109715		<0.002	<0.02	D2 109755		<0.002	<0.02

DKM

Bondar-Clegg & Company Ltd.
 Pemberton Ave.
 North Vancouver, B.C.
 V1P 2R5
 (604) 985-0681 Telex 04-352667



Certificate
 of Analysis

ENTERED
 Apr. 7/87

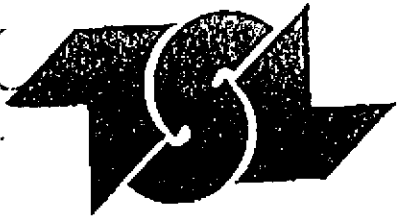
REPORT: V89-01062.4

PROJECT: UNUK

PAGE 4

SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT	SAMPLE NUMBER	ELEMENT UNITS	Au OPT	Ag OPT
D2 109756		<0.002	<0.02				
D2 109757		<0.002	<0.02				
D2 109758		<0.002	<0.02				
D2 109759		<0.002	<0.02				
D2 109760		<0.002	<0.02				
D2 109761		<0.002	<0.02				
D2 109762		<0.002	<0.02				
D2 109763		<0.002	<0.02				
D2 109764		<0.002	0.02				
D2 109765		<0.002	<0.02				
D2 109766		<0.002	<0.02				
D2 109767		<0.002	<0.02				
D2 109768		<0.002	<0.02				
D2 109769		<0.002	<0.02				
D2 109770		<0.002	<0.02				
D2 109771		<0.002	<0.02				
D2 109772		<0.002	<0.02				
D2 109773		<0.002	<0.02				
D2 109774		<0.002	<0.02				
D2 109775		<0.002	<0.02				
D2 109776		<0.002	<0.02				
D2 109777		<0.002	<0.02				
D2 109778		<0.002	<0.02				
D2 109779		<0.002	<0.02				
D2 109780		<0.002	<0.02				
D2 109781		<0.002	<0.02				
D2 109782		<0.002	<0.02				
D2 109783		<0.002	<0.02				
D2 109784		<0.002	<0.02				
D2 109785		<0.002	<0.02				
D2 109786		<0.002	<0.02				
D2 109787		<0.002	<0.02				
D2 109788		<0.002	<0.02				
D2 109789		<0.002	<0.02				
D2 109790		<0.002	<0.02				
D2 109791		<0.002	<0.02				

CA-41



ENTERED APR 24 1989

TSL LABORATORIES

DIV BURGESS TECHNICAL ENTERPRISES LIMITED

2-302-48th STREET, EAST
SASKATOON, SASKATCHEWAN
S7K 6A4

☎ (306) 931-1033 FAX: (306) 242-4717

CERTIFICATE OF ANALYSIS

SAMPLE(S) FROM Prime Exploration Ltd.
10th Floor-Box 10, 808 West Hastings St.
Vancouver, B.C.
V6C 2X6

REPORT No.
S6542

INVOICE #: 11253
P.O.:

SAMPLE(S) OF Pulps

Recheck

Bondar-Clegg

Au
ozt

Ag
ozt

V89-01062.4-109680	<.001	<.05
V89-01062.4-109695	.004/.005	.08
V89-01062.4-109710	<.001	<.05
V89-01062.4-109725	<.001	<.05
V89-01062.4-109740	<.001	<.05
	CA-41	
V89-01062.4-109755	<.001	<.05
V89-01062.4-109770	<.001	<.05
V89-01062.4-109785	<.001	<.05

APPENDIX V

Statement of Expenditures

APPENDIX V

Statement of Expenditures

March 5 to 25, 1989

Personnel

B.W. Downing, geologist, 5 days @ \$300/day	\$1500
G. McArthur, geologist, 5 days @ \$300/day	1500
C. Swanson, geologist, 5 days @ \$200/day	<u>1000</u>
	\$ 4,000.00

Drilling

2667.3 feet (813 metres) NQ core @ \$23.80/foot	62,681.55
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Assays

452 samples for Au, Ag assay sample preparation @ \$16.50/sample	<u>7,458.00</u>
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TOTAL	\$74,139.55
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