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Cascadia International Ventures Inc.

### 1997 DRILLING PROGRAM ON THE FAWN 1-7 CLAIMS

Volume I - Text

Located on the Nechako Plateau Omineca Mining Division NTS 93F/3E 53° 12' North Latitude 125° 08' West Longitude

-prepared for-

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October, 1997 uity Engineering Ltd

### SUMMARY

The Fawn property covers 140 contiguous claim units (3500 hectares) in the Nechako Plateau of central British Columbia, approximately 120 kilometres southwest of Vanderhoof and 180 kilometres west of Quesnel. The Kluskus-Malaput Forest Road passes through the claim group, along with a network of tributary roads and logging spurs. Topography is moderate and lower elevations have been partially clear-cut.

BP Minerals Limited carried out extensive soil sampling over the Fawn property in 1982 and 1983, identifying a northwesterly trend of coincident lead+zinc+silver+arsenic anomalies over 700 x 3,000 metres. Prospecting, mapping and geophysical surveying by Western Keltic Mines Inc. in 1991 showed these anomalies to correspond to four east-west VLF-EM conductors with a minimum aggregate length of 6,500 metres. A hand trench over one of these conductors (V2) revealed sericite-clay alteration with epithermal quartz stockwork which averaged 0.6 g/tonne Au across a true width of 8.2 metres in the Giver Zone. In 1994, Western Keltic drilled 617 metres in widely spaced reconnaissance holes, returning intersections of 8.1 metres of 2.0 g/tonne Au and 4.4 metres of 1.5 g/tonne Au from the Giver Zone. In March and April of 1997, Cascadia International Ventures Inc. drilled a further 620 metres on the Fawn property, consisting of five 150-metre stepouts along the V2 conductor from the Giver Zone and two holes on the Giver Splay, a weakly conductive cross-structure nearby.

The Fawn property is underlain by a sequence of Early to Middle Jurassic Hazelton Group rhyolitic and andesitic volcanics with lesser epiclastic sediments. These have been intruded by a Late Cretaceous diorite and by later felsic dykes. Drilling shows that VLF-EM conductor V2 is caused by a major, steeply north-dipping, fault zone enveloped by 15-30 metres of intense clay-sericite alteration with local chalcedonic quartz stockwork and chalcedony breccias. Sulphide content is variable, consisting mainly of pyrite, lesser arsenopyrite and minor chalcopyrite, sphalerite and galena. Cretaceous or Tertiary quartz-feldspar rhyolite dykes are locally emplaced along the V2 fault zone, as well as faulted and brecciated by it. There is evidence of repeated faulting, dyke emplacement and hydrothermal activity along the fault zones, with several generations of veining and brecciation.

To date, 900 metres of conductor V2 have been tested by drilling, with each of the eight holes intersecting wide clay-sericite alteration zones, variable amounts of quartz veining and gold values in the range of 100-3000 ppb. The best 1997 hole intersected 10.2 metres grading 1.08 g/tonne Au. The remaining 1000+ metres of conductor V2 have not been drilled, nor have the 4,600+ metres along three parallel conductors, although each is accompanied by soil geochemical anomalies of similar intensity to that associated with V2. In addition, it is possible that the fault/alteration zones may be more focused, with narrower widths of better mineralization, where they pass eastwards out of andesitic lapilli tuffs into rhyolite. An analogue for this would be the Tommy epithermal deposit, 17 kilometres to the south, where similar Hazelton Group rhyolites host 478,000 tonnes grading 8.7 g/tonne Au across an average width of four metres.

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### **1997 DRILLING PROGRAM ON THE FAWN 1-7 CLAIMS**

### TABLE OF CONTENTS

		Page
	SUMMARY	. <u>i.</u>
1.0	INTRODUCTION	.1.
2.0	LIST OF CLAIMS	.1.
3.0	LOCATION, ACCESS AND GEOGRAPHY	.1.
4.0	REGIONAL AND PROPERTY EXPLORATION HISTORY	
	4.1 Previous Work	.2.
	4.2 1997 Diamond Drilling Program	.4.
5.0	REGIONAL GEOLOGY	.4.
6.0	PROPERTY GEOLOGY	.6.
7.0	DIAMOND DRILLING	.6.
8.0	DISCUSSION	.11.

### APPENDICES

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Appendix A	Bibliography
Appendix B	Statement of Expenditures
Appendix C	Diamond Drill Logs
Appendix D	Certificates of Analysis
Appendix E	Geologist's and Engineer's Certificates

### LIST OF TABLES

	<u>Page</u>
Claim Data	.1.
Drill Hole Survey Data	.7.
Significant Intercepts - FWN97-01	.7.
Significant Intercepts - FWN97-02	.8.
Significant Intercepts - FWN97-03	.8.
Significant Intercepts - FWN97-04	.9.
Significant Intercepts - FWN97-05	.9.
Significant Intercepts - FWN97-06	.10.
Significant Intercepts - FWN97-07	.11.
	Drill Hole Survey Data Significant Intercepts - FWN97-01 Significant Intercepts - FWN97-02 Significant Intercepts - FWN97-03 Significant Intercepts - FWN97-04 Significant Intercepts - FWN97-05 Significant Intercepts - FWN97-06

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### LIST OF FIGURES

ł

i

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LIST OF FIG	GURES	Following
		Page
Figure 1	Location Map	.1.
Figure 2	Claim Map	.1.
Figure 3	Regional Geology	.4.
Figure 4	Compilation Map	-Pocket-
Figure 5	Drill Plan - Giver Zone	.6.
Figure 6	FWN 94-02, 94-03, 97-01, 97-02 Cross-Section	-Pocket-
Figure 7	FWN 97-02 Cross-Section	-Pocket-
Figure 8	FWN 97-03 Cross-Section	-Pocket-
Figure 9	FWN 97-04 Cross-Section	-Pocket-
Figure 10	FWN 97-05 Cross-Section	-Pocket-
Figure 11	FWN 97-06 Cross-Section	-Pocket-
Figure 12	FWN 97-07 Cross-Section	-Pocket-

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### 1.0 INTRODUCTION

The Fawn property is located on the Nechako Plateau, approximately 120 kilometres southwest of Vanderhoof in central British Columbia. It is underlain by felsic and andesitic Hazelton Group volcanosedimentary rocks cut by the Late Cretaceous Capoose Lake Batholith and by feeder dykes to the Eocene Ootsa Lake Group felsic to andesitic volcanics. BP Minerals Ltd. carried out geological mapping, soil sampling and backhoe trenching on the property from 1981 to 1984, defining coincident zinc-silver-lead soil anomalies over an area of 3000 metres by 700 metres. It was restaked as the Fawn property and Western Keltic Mines Inc. conducted mapping, prospecting, soil sampling, geophysical surveys and 617 metres of diamond drilling from 1991 through 1994. Four open-ended, subparallel VLF-EM conductors, with a total strike length of 6,400 metres, were defined within the soil geochemical anomaly. Drilling on one of these, the Giver Zone, showed it to correspond to epithermal chalcedony stockwork/breccia within a 18-32 metre wide zone sericite-clay alteration; the best intersection assayed 2.0 g/t Au across 8.1 metres.

A 620 metre diamond drilling program was carried out in March and April of 1997, to intersect the Giver Zone conductor along strike from the 1994 drilling and to test one of its splays which had yielded auriferous subcrop mineralization. Equity Engineering Ltd. conducted this drill program for Cascadia International Ventures Inc. and has been retained to report on the fieldwork. Limited mapping and soil sampling were carried out in conjunction with the drillsite reclamation in September 1997 and will be reported separately.

### 2.0 LIST OF CLAIMS

The Fawn property comprises seven contiguous claims totalling 140 claim units (3,500 hectares), located in the Omineca Mining Division (Figure 2). Records of the British Columbia Energy and Minerals Division indicate that the Fawn 1-7 claims are owned by Western Keltic Mines Inc.. Separate documents indicate that Cascadia International Ventures Inc. has an option to earn an interest in them. Claim data for the Fawn property is summarized in Table 2.0.1.

### <u>Table 2.0.1</u> CLAIM DATA

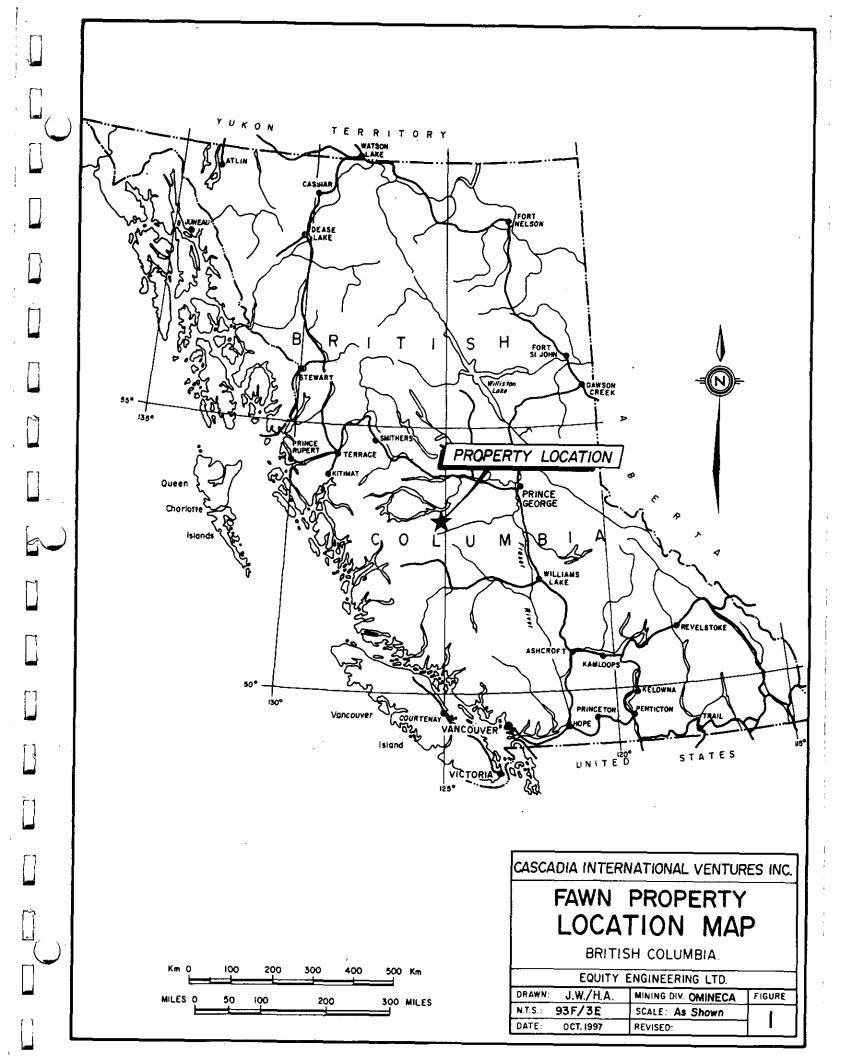
Claim Name	Mineral Tenure No.	No. of Units	Record Date	Expiry Year
Fawn 1	243221	20	March 15, 1991	2007*
Fawn 2	301430	20	June 26, 1991	2007*
Fawn 3	301431	20	June 26, 1991	2007*
Fawn 4	301432	20	June 26, 1991	2004
Fawn 5	305450	20	October 13, 1991	2007*
Fawn 6	322193	20	October 28, 1993	2007*
Fawn 7	323869	20	February 26, 1994	2004
		140		

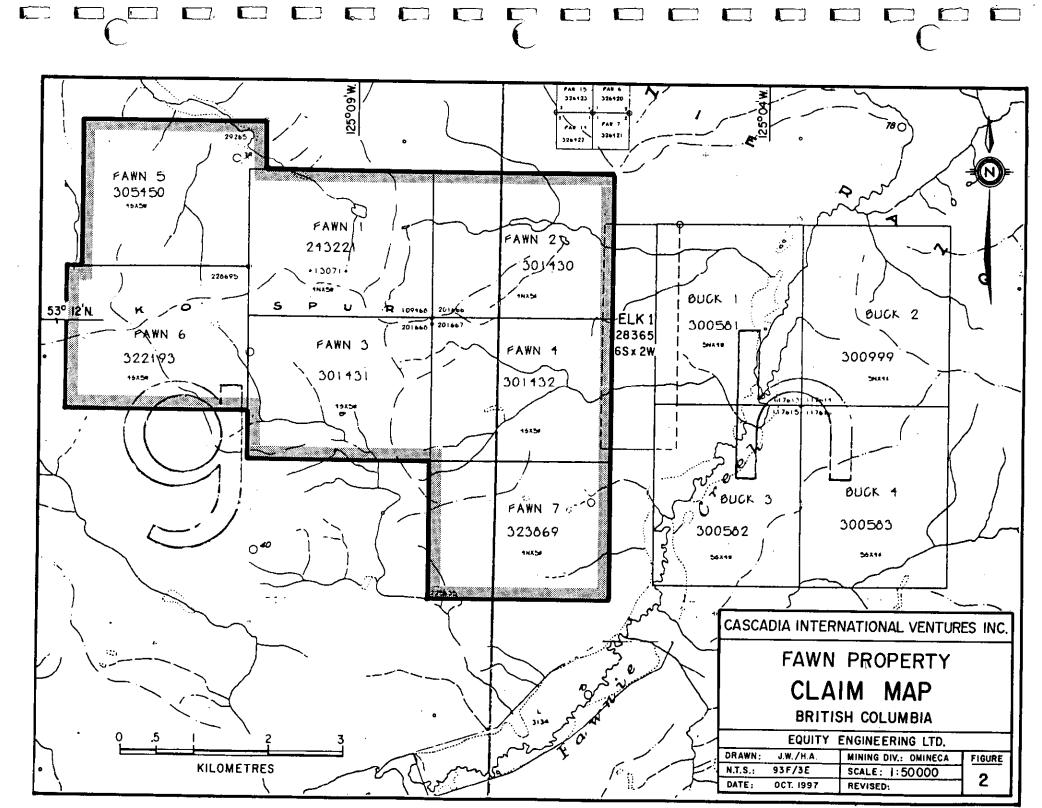
\* Subject to approval of assessment work covered by this report.

The position of the legal corner posts for the Fawn 1-5 claims has been verified by the authors.

### 3.0 LOCATION, ACCESS AND GEOGRAPHY

The Fawn property is situated on the Nechako Plateau of central British Columbia, approximately 120 kilometres southwest of Vanderhoof and 180 kilometres west of Quesnel (Figure 1). The claims are





located within the Omineca Mining Division, centred at 53° 12' north latitude and 125° 08' west longitude.

The property is accessed by a major logging road, the Kluskus-Malaput Forest Road, which reaches the north side of the property 146 kilometres south of the Plateau Forest Products mill at Engen on Highway 16. The Kluskus-Malaput road angles through the southeastern corner of the property, while a major branch, the Van Tine Forest Road, provides good access through its northern part. The M-4000 Forest Road, another major branch, leaves the Kluskus-Malaput south of the property and angles northwesterly through the southwestern corner of the Fawn 6 claim. Spur roads provide four-wheel drive access throughout each of several recent clear-cuts on the property. The Capoose access road, on the north side of Van Tine Creek, is also accessible by four-wheel drive vehicle but has not been maintained for several years.

The claims cover the eastern portion of Entiako Spur, a range of rolling hills lying south of Van Tine Creek within the Nechako Plateau. Upland surfaces are generally well drained with few lakes or marshes. Lower creek valleys are broad and swampy. Topography is moderate, with elevations ranging from 1,100 metres in the Fawnie Creek valley to 1,739 metres at the highest point of Entiako Spur. Outcrop exposure is fairly good along the ridge top, but is increasingly masked by glacial till at lower elevations. Overall, the property would average less than 5% outcrop. Road cuts along the Van Tine Road expose up to 30 metres of glacial till. Glacial striae trend 060° on the Fawn 2 claim, and Tipper (1963) provides strong evidence regionally for a southwestern ice source.

The property is largely covered by spruce and lodgepole pine with a light undergrowth of huckleberry and alder. Recent clear-cuts at lower elevations on most of the claims have made the sparse outcrops easier to find and examine. The Fawn property is subject to a continental climatic regime, with warm summers and cold winters. Snowfall is moderate with an accumulation of one to two metres during the winter.

### 4.0 REGIONAL AND PROPERTY EXPLORATION HISTORY

### 4.1 Previous Work

The area around the Fawn property received little exploration until the late 1960's, when Rio Tinto Canadian Exploration Ltd. carried out stream and lake sediment sampling surveys throughout the Nechako Plateau, searching primarily for copper-molybdenum porphyry deposits (Hoffman, 1976). Follow-up work on one of their anomalies by Rio Canex (1969-71) and Granges Exploration Ltd./Cominco Ltd. (1976-present) led to the discovery in 1979 of the Capoose silver-lead-zinc deposit approximately seven kilometres north of the Fawn property. Reserves at Capoose have been estimated at 28 million tonnes grading 36 g/tonne silver and 0.9 g/tonne gold (Green and Diakow, 1993).

Following the recognition of a major silver resource at Capoose, BP Minerals Limited staked several other nearby high-priority silver-lead-zinc lake sediment anomalies from Rio Canex's data. Their Gran and Laid claims were staked in 1981 to cover the drainages surrounding Square Lake, a small lake at the head of Van Tine Creek near the northern boundary of the present Fawn 1 claim. Square Lake was extremely anomalous in lead, exceeding the values for the lakes which marked the Capoose deposit (Hoffman, 1976).

In 1982, BP Minerals carried out geological mapping over the area now covered by the Fawn property and laid out a compass and hipchain geochemical grid which used three different numbering systems. An east-west baseline was blazed and numbered from 0+00W to 28+00W, just north of the present Fawn 2 southern boundary. Cross-lines were run to the south from this baseline, with station numbering up to 24+00S. A second baseline was blazed to the north from station 28+00W on the first baseline, which was re-labelled 0+00N 0+00W. Cross-lines were run to the east and west from this second baseline (and labelled accordingly), which extended north to 18+00N. A western tie line was

blazed north-south 2,600 metres to the west of the second baseline, near the western boundary of the current Fawn 1 and 3 claims. This was used to tie in lines 0+00N to 14+00N, which were run west from the second baseline. Lines were also run and numbered east (Lines 14+00N to 20+00N) and west from the western tie line (and labelled east or west relative to the western tie line). A total of 1,152 soil and stream sediment samples were taken in 1982 and a further 1,517 in 1983 from ground currently covered by the Fawn property (Hoffman and Smith, 1982; Smith and Hoffman, 1983 and 1984). Samples were taken initially at 100 metre intervals on lines spaced 100 metres apart, with later infilling to 50 metre intervals in anomalous areas. The soil geochemistry delineated a northwesterly trend of coincident lead-zinc-silver anomalies measuring approximately 3,000 metres by 700 metres, centred on the Fawn 1 claim.

In 1983, limited trenching and a series of 40 backhoe test pits were excavated at 25 metre intervals near the eastern end of the lead-zinc-silver soil anomaly, exposing three or four "rhyodacite lapilli tuff" units with up to 94.5 ppm silver and 880 ppb gold (Smith and Hoffman, 1984). The following year, another grid was established for mapping purposes over the Fawn 1 soil anomaly. A 3,000 metre baseline oriented at 310° was cut and numbered from 0+00N to 30+00N. Cross-lines were run at 035° from the baseline at 200 metre intervals. Further backhoe trenching was carried out in the area of the 1983 trenching and near the western end of the soil anomaly, without encouraging results (Smith, 1985). BP Minerals allowed their claims to lapse in 1988.

The Fawn 1-4 claims were staked in 1991 over BP Minerals' soil geochemical anomaly. In September and October of that year, Western Keltic Mines Inc. carried out geological mapping, soil and rock geochemistry and geophysical surveying, taking 239 rock, 144 soil and 41 deep overburden samples. The 1984 cut baseline was re-established and extended at 130° for 2,425 metres to the southeast. Cross-lines were run towards 040° at 100 metre intervals from 4+00N to 30+00N and at 200 metre intervals from 4+00N to 24+00S, with stations marked every 25 metres. Cross-lines, 500 metres in length, were run at a bearing of 220° from 5+00N to 27+00N at 100 metre intervals. Five widely-spaced lines were extended further to the southwest, in an area to the south of pre-existing coverage and soil samples were taken along them at 50 metre intervals. The BP Minerals soil anomalies were relocated relative to the new grid and verified by 41 soil samples taken from their most anomalous sample locations. Magnetometer and VLF-EM surveys were carried out over 31 line-kilometres of the grid between 2+00S and 30+00N. Deep overburden sampling and MaxMin EM were tested over the Giver Zone, a mineralized VLF-EM conductor (Awmack, 1991).

Four subparallel, easterly-trending VLF-EM conductors were defined along strike lengths of 700 to 2200 metres by the 1991 program, with each remaining open along strike in at least one direction. Each of the four VLF conductors is accompanied by silver+zinc+lead+arsenic soil geochemistry. Eocene(?) epithermal chalcedony-sulphide breccia was found in subcrop and float along one of the VLF conductors, with assays up to 12.9 g/tonne Au and 637 g/tonne silver in separate samples from the "Giver Zone" and one of its splays, the "Giver Splay" (Awmack, 1991). The Fawn 5 and 6 claims were subsequently staked to cover the projected westward extension of these VLF structures.

Western Keltic performed a 20.7 line-kilometre induced polarization survey on lines spaced 200 metres apart from 3+00N to 29+00N in October and November, 1993. This showed low resistivity and weak chargeability along the Giver VLF-EM structure and outlined a strong chargeability anomaly at the eastern end of the survey. Moderate chargeability and low resistivity anomalies were indicated near the northwestern end of the grid, in an area of strong soil geochemistry and two VLF-EM structures (Ballantyne, 1993).

During the course of regional mapping in 1993, the BC Geological Survey discovered the Malaput Showing, a zone of silicification and sericitization located four kilometres southeast of the Giver Zone (Diakow and Webster, 1994). The Fawn 7 claim was subsequently staked over the Malaput Showing.

The BC Geological Survey undertook regional lake sediment (Cook and Jackaman, 1994) and

basal till (Levson et al, 1994) sampling programs throughout portions of the 93F map sheet in 1993, taking three lake sediment samples and 18 till samples from the Fawn property. The lake sediment sample from Square Lake returned the highest lead, zinc and cobalt values for all 237 samples taken from the region, along with anomalous antimony, arsenic and gold. Six of the till samples exceeded the survey's 95th percentile for gold, lead, arsenic or antimony. Four of these anomalous till samples, including the sample with the survey's second highest gold value, were taken from the northeastern portion of the Fawn 7 claim, an area which has received no exploration.

In 1994, Western Keltic drilled 617 metres in six diamond drill holes on geophysical and geochemical targets on the Fawn 1 and Fawn 5 claims. Three of these were drilled across the V2 conductor (Giver Zone) showing it to be a steeply-dipping 18-32 metre wide zone of sericite-clay-pyrite alteration hosting epithermal chalcedony stockworks and breccias. The best intersections were 8.1 metres of 2.0 g/tonne Au in hole FWN94-02 and 4.4 metres of 1.5 g/tonne Au and 63.8 g/tonne Ag in hole FWN94-03. In addition, 55 soil samples were taken from a small grid over the Malaput Showing, returning up to 255 ppb Au, 336 ppm As, 226 ppm Pb and 1360 ppm Zn. Mapping showed it to be an easterly-trending, 25-30 metre wide zone of silicification which can be traced along strike for at least 300 metres (Baknes and Awmack, 1994a).

### 4.2 1997 Diamond Drilling Program

During March and April of 1997, Cascadia International Ventures Inc. carried out a second diamond drill program on the Fawn property, targeted at the Giver Zone and Giver Splay on the Fawn 1 claim. Seven holes, totalling 619.6 metres (2,032') of BTW core, were drilled by Falcon Drilling Ltd. of Prince George, using their F-1000 drill. Core was logged on site and split mechanically for geochemical analysis. Drill logs are attached in Appendix C. Drill sites were accessed by short spur roads built using a small cat from existing roads. A total of 125 core samples were analyzed geochemically for gold and by ICP for 32 elements by Chemex Laboratories in North Vancouver. Samples with high gold values were subsequently fire assayed. Appendix D contains analytical certificates.

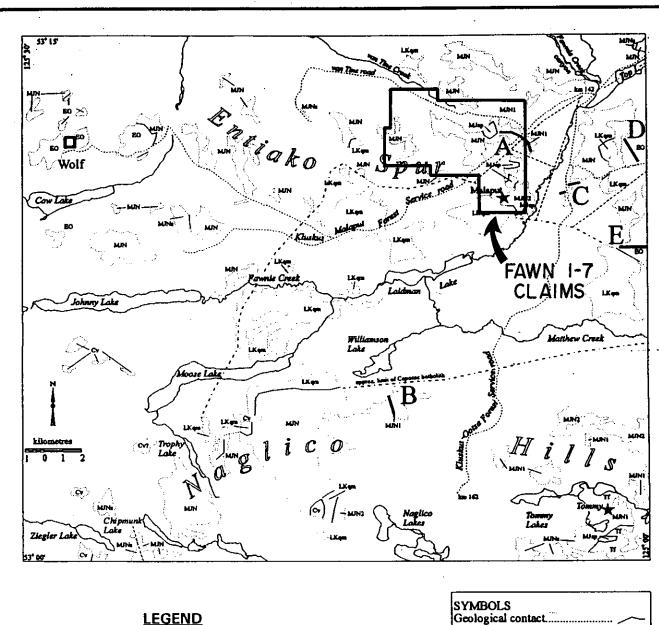
Reclamation of all drill sites and drill roads was carried out in September 1997. Limited soil sampling and mapping was performed in conjunction with the reclamation, focused on the Malaput Showing area and skarn mineralization on the Fawn 6 claim. Results of this program will be reported separately.

### 5.0 REGIONAL GEOLOGY

The British Columbia Geological Survey carried out 1:50,000 scale regional mapping over mapsheet 93F/6 in 1992 (Green and Diakow, 1993; Diakow and Green, 1993). In 1993, this mapping was extended to the south over map-sheet 93F/3, which covers the Fawn property (Diakow and Webster, 1994; Diakow et al, 1994). Their mapping shows Jurassic Hazelton Group volcanics and sediments intruded by the Late Cretaceous Capoose Lake batholith and unconformably overlain by Eocene Ootsa Lake Group subaerial volcanics and younger plateau basalts (Figure 3).

The Early to Middle Jurassic Hazelton Group rocks in the vicinity of the Fawn property have been assigned by Diakow and Webster (1994) to their informal Naglico Formation of silica-bimodal volcanic rocks and Bajocian intravolcanic sediments which are gradationally overlain by Callovian marine sediments. The lower division of this formation consists of "crudely layered fragmental and lesser flow rocks of rhyolitic composition, and local maroon and green andesitic tuffs deposited in a subaerial environment" (**Unit MJN1**). The upper division is dominated by mafic and intermediate lavas (**Unit MJN2**), interpreted by Diakow and Webster (1994, p. 19) to be deposited in a shallow marine environment with local subaerial conditions. Green and Diakow (1993) report that a section of the upper division exceeds 1,000 metres in thickness on Tutiai Mountain, 14 kilometres north of the Fawn property. Augite porphyry plugs (**Unit MJap**) mapped on the Fawn claims are thought to be cogenetic with upper

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#### STRATIFIED ROCKS MIOCENE TO PLIOCENE

Chilcotin Group Cv Olivine basalt EOCENE Ootsa Lake Group EO Rhyolite and andesite flows, quartzbearing lapilli tuffs, tuffaceous

siltstone MIDDLE JURASSIC Hazelton Group (Naglico Formation) MJNs Fine to coarse-grained, fossiliferous volcaniclastics MJN2 Basalt and andesite flows and lapilli tuffs

MJN1 Rhyolite flows, ash-flow tuffs and lapilli tuffs

## INTRUSIVE ROCKS

Felsite sills LATE CRETACEOUS

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Capoose Lake Batholith LKqm Equigranular quartz monzonite, with lesser quartz diorite and quartz porphyry MIDDLE JURASSIC

10

MJap Mafic augite-plagioclase porphyry plugs

Geology modified from Diakow and Webster (1994).

KILOMETRES

1	CASCADIA INTERNATIONAL VENTURES INC.												
⊫		FAWN	PROPERTY										
	REGIONAL GEOLOGY												
	BRITISH COLUMBIA												
		EQUITY	ENGINEERING LTD.										
	DRAWN:	J.W. / H.A.	MINING DIV.: OMINECA	FIGURE									
	N.T.S.:	93F/3	SCALE : 1:200,000	3									
	DATE:	OCT. 1997	REVISED	Э									

division Naglico Formation augite-phyric volcanics.

Wide-spread, irregularly-distributed, marine sedimentary rocks (**Unit MJNs**) are intercalated with Naglico Formation volcanics, interpreted as basins between coalescing volcanic centres. The marine sediments become dominant in the stratigraphically highest Middle Jurassic exposures. Main lithologies include feldspathic sandstone and siltstone, tuffaceous argillite, locally prominent volcanic conglomerate and scarce limestone. Fossils are common in the sedimentary rocks, with most of indeterminate or probable Middle Jurassic age and at least one early Bajocian collection (Diakow and Webster, 1994).

The Jurassic stratigraphy was intruded by the Late Cretaceous Capoose Lake Batholith (**Unit LKqm**), a 250 km<sup>2</sup> pluton which extends southwesterly for at least 20 kilometres from the Fawn property. The Hazelton volcanics of the southwestern portion of the Fawn property are thought to be underlain by the Capoose Lake Batholith at a fairly shallow depth. Its main phase consists of light coloured, medium-to coarse-grained, equigranular quartz monzonite, although its composition is locally granodioritic and a dioritic phase cuts northerly through the Fawn 2, 4 and 7 claims. Andrew (1988) reports a biotite K-Ar date of 64.3+2.4 Ma for the batholith. Miarolytic quartz porphyry dykes and plugs cut Hazelton Group sediments on the Buck property, immediately east of the Fawn claims. These were interpreted by Diakow and Webster (1994) to be subvolcanic apophyses projecting from the Capoose Lake Batholith.

Flat-lying to moderately dipping, subaerial volcanics of the Ootsa Lake Group (Unit EO) unconformably overlie older Mesozoic rocks. Potassium-argon dating of Ootsa Lake rocks at the Wolf prospect gave an age of 48±2 million years (mid-Eocene). The Ootsa Lake volcanics consist of calcalkaline andesite to rhyolite. North of the Natalkuz Fault, a northeasterly trending fault which passes twenty kilometres northwest of the Fawn claims, Ootsa Lake volcanics cover an extensive area, with a 750 metre stratigraphic section. South of the fault, the Ootsa Lake Group forms thin isolated cappings on older rocks.

Miocene plateau basalts of the Chilcotin Group (Unit Cv) unconformably overlie all other units.

Low grade regional metamorphism and weak deformation are pervasive on the Nechako Plateau. Contact metamorphism is pronounced around intrusives. Tipper (1959) observed that the overall lack of structural features may, in part, be attributed to the abundance of often structureless volcanics in the area. The Hazelton volcanics appear more strongly deformed in comparison to other rock types, with dips of up to 70°. At the Capoose deposit, eight kilometres north of the Fawn property, bedding dips moderately (20-40°) to the southwest, with a synclinal fold axis plunging at 10° to the southeast (Andrew and Godwin, 1987). The Ootsa Lake Group volcanics were deposited in a period of extensional tectonism. Another period of deformation during the Oligocene produced broad open folds in the Ootsa Lake Group volcanics and sediments. The relatively undeformed Chilcotin Group consists of generally flat-lying to gently easterly dipping plateau lavas (Tipper, 1963).

Several styles and ages of mineralization have been documented in the vicinity of the Fawn property (Figure 3). Teck Corp.'s Tommy epithermal gold-silver prospect, 17 kilometres south of the Fawn claims, consists of several north to northeast trending veins and silicified stockwork zones hosted by Naglico Formation quartz-phyric felsic crystal lithic and ash tuffs. The veins consist of milky quartz, chalcedony, sparry calcite, ankerite and adularia, with typical epithermal textures such as druse, colloform banding, cockscomb structures and multiple brecciation/veining episodes. Only trace amounts of sulphides, mainly pyrite, chalcopyrite, sphalerite and galena, are present. The Tommy Vein, which has received the most exploration, hosts a geological resource of 478,000 tonnes grading 8.7 g/tonne Au and 82.3 g/tonne Ag across an average width of four metres (J. Pautler, pers. comm., 1997).

The Wolf epithermal gold-silver prospect, located twenty kilometres west of the Fawn property, is hosted by Eocene Ootsa Lake Group rhyolitic flows, tuffs and subvolcanic intrusives. Repeated low-sulphide silicification, brecciation and stockwork veining have been accompanied by up to 8.49 g/tonne gold and 42.2 g/tonne silver across 7.5 metres in trenching (Cann, 1984). It has been suggested that

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the Wolf deposit may have been related to maar (Andrew et al, 1986), collapse caldera (Andrew, 1988) or hot-spring (Andrew, 1988) paleo-environments.

The Capoose silver deposit, located eight kilometres north of the Fawn property, is hosted by Naglico Formation mafic flows, rhyolite tuff, argillite and lithic wacke intruded by Late Cretaceous quartz-garnet rhyolite sills related to the Capoose Lake Batholith. Mineralization consists of pyrite, sphalerite, galena, chalcopyrite and arsenopyrite in disseminations, fracture-fillings and replacing garnets, and is thought to be Late Cretaceous in age (Andrew, 1988). The Capoose deposit contains 28 million tonnes grading 36 g/tonne silver and 0.9 g/tonne gold (Green and Diakow, 1993). The Capoose Lake Batholith itself has been explored for porphyry-style copper-molybdenum mineralization a few kilometres west of the Capoose deposit.

Immediately east of the Fawn property, the Buck 1-4 claims cover a 3,000 metre long zincarsenic-lead soil geochemical anomaly overlying Naglico Formation rocks. Proximal (vent facies) felsic volcanics change laterally to distal felsic volcaniclastics and epiclastics along with marine sedimentary and intermediate volcanic lithologies. Stratabound sphalerite-pyrrhotite mineralization, grading up to 4.69% zinc, is present in felsic ash tuffs. The Christmas Cake Showing, with a 45 centimetre chip sample grading 7.38% Zn, 2.25% Pb and 542 g/tonne Au, consists of coarse sphalerite, iron carbonate, galena, minor chalcopyrite and sugary quartz forming a matrix which supports fragments composed entirely of very fine-grained pyrite and by variably altered, angular, felsic lithic clasts (Baknes and Awmack, 1994). A northeast-trending VLF-EM conductor corresponds to a recessive zone of clay alteration with quartz-calcite veining, accompanied by 2-10% pyrite and lesser arsenopyrite and sphalerite. Although this zone returned only low gold and silver values, its similarities to the Fawn's Giver Zone suggest a genetic link (Caulfield, 1996)..

Fifteen kilometres east of the Fawn property, the PEM prospect is underlain by Naglico Formation felsic to intermediate tuffs, lapilli tuffs, breccias and flows, intercalated with argillite, siltstone and sandstone. Disseminated and shear-hosted mineralization occurs in a steeply-dipping, structurally-controlled zone of phyllic and argillic alteration at least 900 metres long, with introduction of 3-4% sphalerite and 1-2% pyrite (Schroeter and Lane, 1994). Zbitnoff (1988) reports drill intersections up to 6.3 metres grading 14.3 g/tonne gold, 27 g/tonne silver and 1.25% zinc. Textural evidence suggests that PEM mineralization may be genetically similar to that of Capoose.

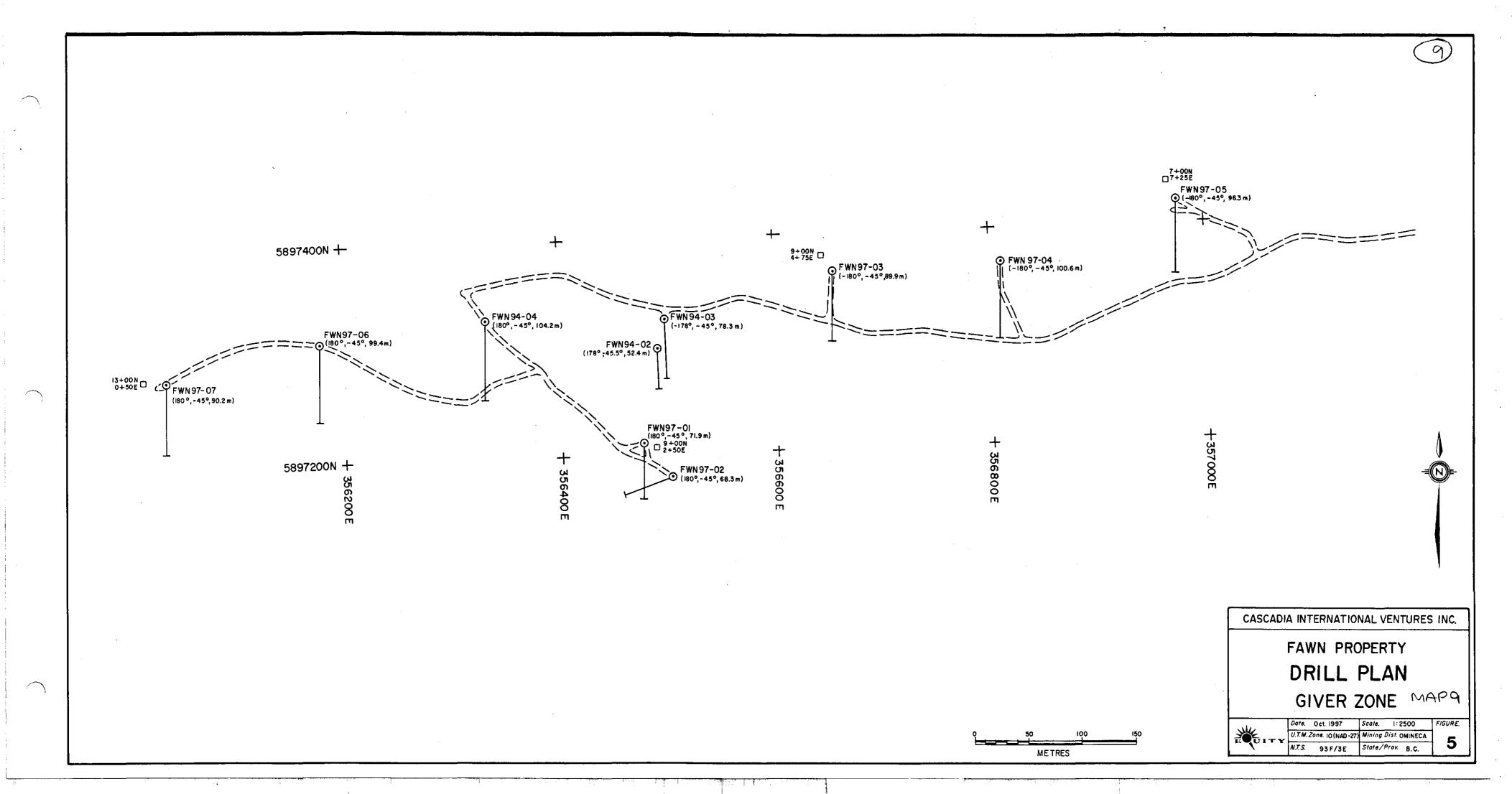
### 6.0 **PROPERTY GEOLOGY**

The Fawn property is largely underlain by a sequence of Early to Middle Jurassic Hazelton Group (Naglico Formation) rhyolitic and andesitic volcanics with lesser epiclastic sediments. These have been intruded by a dioritic pluton, thought to form part of the Late Cretaceous Capoose Lake Batholith, and by later felsic dykes which are presumably feeders to the Tertiary Ootsa Lake rhyolites. No geological mapping was carried out on the Fawn property during the 1997 drill program; more detailed descriptions of geology and mineralization can be found in previous reports by Baknes and Awmack (1994a) and Awmack (1991).

### 7.0 DIAMOND DRILLING

Seven holes were drilled in the vicinity of the gold-bearing 1994 drill intersections on the Giver Zone. Five of these were 150-metre stepouts along the V2 (Giver Zone) VLF-EM conductor from holes FWN94-02, 03 and 04. The other two were designed to test the Giver Splay, a conductive cross-structure where epithermal veining had assayed up to 12.9 g/tonne Au in subcrop. Table 7.0.1 summarizes location, orientation and drilling depths for the 1997 holes. The holes are located in plan on Figures 4 and 5, with vertical cross-sections in Figures 6-12. Drill logs are attached in Appendix C.

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Hole	Grid Lo	ocation	Elevation	Azimuth	Dip	Length
Number	Northing (m)	Easting (m)	(m)	(°)	(°)	(m)
FWN97-01	9+12 N	2+45 E	1525.7	180	-45	71.9
FWN97-02	8+72 N	2+40 E	1528.7	250	-45	71.3
FWN97-03	8+82 N	4+71 E	1478.3	180	-45	89.9
FWN97-04	7+68 N	5+70 E	1457.2	180	-45	100.6
FWN97-05	6+84 N	7+18 E	1425.3	180	-45	96.3
FWN97-06	11+98 N	1+81E	1574.8	180	-45	99.4
FWN97-07	12+87 N	0+62 E	1598.5	180	-45	90.2
				-		619.6

### Table 7.0.1 Drill Hole Survey Data

### FWN97-01

Hole FWN97-01 was drilled under the Giver Splay along the same section as 1994 drill holes FWN94-02 and -03. The Giver Splay is a weak VLF-EM conductor which trends 130° from conductor V2 (Giver Zone). In 1991, a sample of silicified, brecciated, pyritic subcrop cut by drusy quartz-arsenopyrite veinlets assayed 12.9 g/tonne Au; it was taken from an old backhoe pit a few metres northwest along the Giver Splay from where hole FWN97-01 cuts it. The Giver Splay had been tested previously by FWN94-04, which was targeted at its junction with the Giver Zone conductor, 180 metres to the northwest; it intersected 2.7 metres grading 2.42 g/tonne Au.

The hole was drilled through a package of andesite lapilli to ash tuff which displays zones of alteration (Figure 6). This alteration zoning progresses from propylitic to intensely sericitized andesite lapilli ash tuff. The propylitic alteration displays variable epidote, chlorite and calcite alteration with a significant magnetite content. With increasing alteration the magnetite appears to be altered to hematite with subsequent loss of magnetism and increased red to brown colour. Mixed hematite-sericite alteration then follows which results in mottled red-brown and pale green coloured lapilli-tuff. With increased alteration the colour changes to pale green-grey, then textural features of the lapilli-tuff are obliterated, commonly with increased disseminated pyrite.

FWN97-01 encountered the Giver Splay between 58.8 and 63.0 metres depth. It is a highly fractured zone of intensely sericitized and variably pyritized lapilli tuff, with local quartz+dolomite veining. A strong fracture foliation trends 10-35° to core axis, indicating that the drill hole is oriented at a low angle to the Giver Splay. This is consistent with its assumed northwesterly trend and steep dip. Assuming a 25° average angle, the true width of the Giver Splay would be about 1.8 metres; the true width of sample 197021 (2.02 g/tonne Au) would be 0.46 metres. Significant samples are summarized in Table 7.0.2.

### <u>Table 7.0.2</u> Significant Intercepts - FWN97-01

Sample	From	To	Length	Au	Au	Ag	As	Cu	Pb	Sb	Zn
Number	(m)	(m)	(m)	(ppb)	(g/t)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)
197019	58.80	60.15	1.35	365		2.8	928	27	60	20	440
197020	60.15	61.90	1.75	<5		0.4	32	8	34	6	178
197021	61.90	63.00	1.10	2000	2.02	6.0	2270	9	84	10	508

### FWN97-02

FWN97-02 was directed southwesterly across the same target as FWN97-01, in an effort to cut

the Giver Splay at a less acute angle. Like hole FWN97-01, it was drilled entirely within andesitic lapilli tuff (Figure 7). Strong sericite alteration was observed below the casing from 3.05 to 8.20 metres as well as paralleling a major fault intersected at 48.3 to 52.2 metres and a fault zone from 63.6 to 64.9 metres. The alteration envelopes parallel these fault zones and alteration decreases with distance from the faults. Presumably, the faults correspond to the Giver Splay. Fracturing is generally 50-55° to core axis, corresponding to the Giver Splay's assumed northwesterly trend and steep dip.

# Table 7.0.3 Significant Intercepts - FWN97-02

Sample Number	From (m)	To (m)	Length (m)	Au (ppb)	Au (g/t)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
197024	13.40	14.70	1.30	170		1.2	6	28	262	<2	484
197026	50.00	52.20	2.20	130		3.8	264	28	44	14	154

### FWN97-03

Holes FWN97-03, -04 and -05 were successive 150 metre stepouts to the east along the V2 conductor (Giver Zone) from 1994 holes FWN94-02 and -03. Hole FWN97-03 was collared in a finegrained andesite flow, entering andesitic lapilli tuff at 9.7 metres (Figure 8). It intersected the Giver Zone from 50.5 - 83.6 metres depth, a true width of about 27 metres exhibiting intense sericite+clay alteration and heavy fracturing at 30-65° to core axis. In the core of the Giver Zone, quartz veining increases and two sections of quartz+dolomite vein breccia were intersected, roughly paralleling the core axis (67.6-74.1m and 77.0-81.5m). Slightly elevated gold and arsenic values are associated with the vein breccia (Table 7.0.3).

### Table 7.0.4 Significant Intercepts - FWN97-03

Sample Number	From (m)	To (m)	Length (m)	Au (ppb)	Au (g/t)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
197042	70.41	74.10	3.69	575		2.2	1750	11	68	6	348
197044	77.00	78.90	1.90	345		1.4	1530	12	24	8	186
197045	78.90	81.50	2.60	345		1.8	1420	14	36	8	224

### FWN97-04

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Hole FWN 97-04 was collared 150 metres east of FWN97-03 along VLF-EM conductor V2. It displayed narrow intervals of variably hematite, sericite and clay altered andesite lapilli tuff over the upper 56.1 metres (Figure 9). From 56.1 to 90.4 metres the hole cut a zone of very intensely altered, fractured and faulted lapilli to ash tuff and a minor quartz feldspar porphyry dyke. A very strong fault zone from 59.8 to 65.0 metres displayed strongly brecciated lapilli tuff with clay gouge and alteration. The fault/fracture zone appears to continue to 81.5 metres as all rock units are broken and blocky, commonly with clay on fracture/fault slip surfaces. Broken core within the fault zone display fragments of vein quartz as well as <5mm quartz veins cross-cutting the fault breccia. This suggests multiphase quartz episodes and activity along the V2 conductor.

A quartz stringer zone from 56.3 to 57.2 metres consists of vuggy, generally 1-2 mm (to a maximum of 30 mm) grey quartz stringers with traces of arsenopyrite, chalcopyrite and sphalerite. A later dolomite stringering episode does not appear to host any significant sulphide mineralization. Minor quartz+pyrite+arsenopyrite stringers occur in the 57.2 to 59.8 metre interval and rare quartz + sulphide stringering occur from 63.3 to 79.3 metres.

A quartz-feldspar porphyry dyke intersected from 81.5 to 86.0 metres displays strong fracturing with significant alteration paralleling these fractures suggesting later movement on the fault zone.

Sample	From	То	Length	Au	Au	Ag	As	Cu	Pb	Sb	Zn
Number	(m)	(m)	(m)	(ppb)	(g/t)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)
197049	14.30	15.80	1.50	130		<.2	14	46	2	12	120
197057	56.30	57.20	0.90	125		1.2	282	10	16	8	70
197059	57.60	58.70	1.10	130		3.2	370	27	18	12	176
197061	59.40	59.80	0.40	580		6.6	3690	14	50	14	266
197065	63.30	65.00	1.70	140		1.8	622	11	30	8	184
197067	66.50	67.40	0.90	120		1.2	1200	17	18	10	128
197071	74.80	75.90	1.10	275		2.2	638	17	24	8	284

<u>Table 7.0.5</u> Significant Intercepts - FWN97-04

### FWN97-05

Hole FWN 97-05 was the easternmost stepout along conductor V2, collared another 160 metres east of FWN97-04 in an area of anomalous arsenic soil geochemistry. The upper 50.5 metres cut a variably sericite-hematite altered interval of ash and lapilli tuff (Figure 10). Strong sericite alteration is associated with a fault at 21.9 to 22.4 metres. Minor silicified zones occur in the interval from 20.0 to 21.9 metres above the fault.

Quartz-feldspar porphyry was intersected from 50.5 to 86.6 metres. This interval was subdivided into two units based upon their groundmass colour difference as well as significant fracturing differences. The upper 50.5 to 78.0 metre interval displayed strong fracturing and weak mineralization, with white and grey quartz stringers hosting trace pyrite and arsenopyrite. An extremely rubbly sericite and clay altered zone from 60.5 to 65.9 metres is likely responsible for the V2 conductor.

Gold-bearing mineralization occurs at the footwall contact of the quartz feldspar porphyry. Throughout the mixed porphyry/tuff zone (86.6-89.1 metres) and footwall ash lapilli tuff zone (89.1-91.7 metres), millimetre-scale white and grey quartz stringers host minor amounts of pyrite, arsenopyrite, sphalerite, galena and chalcopyrite. Very fine grained black alteration associated with this zone may be the result of fine-grained sulphides in the wall rock.

Sample Number	From (m)	To (m)	Length (m)	Au (ppb)	Au (g/t)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
197092	86.60	87.90	1.30	575		4.4	1295	8	10	6	74
197093	87.90	89.10	1.20	210		3.0	892	12	12	8	106
197094	89.10	90.60	1.50	1620	1.58	0.2	16	31	62	2	312
197095	90.60	91.70	1.10	300		0.6	62	47	56	6	368

### Table 7.0.6 Significant Intercepts - FWN97-05

### FWN97-06

Holes FWN97-06 and FWN97-07 were stepouts west along conductor V2 from 1994 drill hole FWN94-04, which intersected 2.7 metres assaying 2.4 g/tonne Au. Hole FWN97-06 was collared 150 metres west of FWN94-04 in the vicinity of several epithermal float boulders sampled in 1991. These samples, known as the Givermore Zone, graded up to 3.4 g/tonne Au and 92 g/tonne Ag.

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FWN97-06 encountered dominantly andesite flows with minor andesite lapilli tuffs (Figure 11). Alteration throughout the interval from 43.3 to 71.1 metres is dominantly sericite, clay and pyrite alteration with the most intense alteration paralleling a fault zone from 59.74 to 69.5 metres. Pyrite mineralization is dominant throughout the altered interval and likely represents pyritization of the magnetite (magnetite—hematite—pyrite). Within the fault zone is an interval of silicified breccia with dark grey chalcedonic quartz and minor white quartz from 59.74 to 61.4 metres. On either side of this silicified fault breccia occurs strongly altered and variably quartz and dolomite stringered andesite. Chalcedonic quartz stringers host varying amounts of pyrite, arsenopyrite and sphalerite but sulphides are generally present as trace amounts. These stringers occur sporadically throughout the altered zone, but are also concentrated as a stockwork zone from 53.1 to 55.3 metres. The upper, silicified, part of the fault and the overlying alteration averages 1.08 g/tonne Au and 23 g/tonne Ag along 10.2 metres of core (51.2 - 61.4 metres).

A strong fault breccia with clay gouge intersected from 61.4 to 62.5 metres, and an extremely broken and faulted clay and sericite altered zone from 62.5 to 69.5 metres are thought to be the source of the V2 conductor.

Sample	From	То	Length	Au	Au	Ag	As	Cu	Pb	Sb	Zn
Number	(m)	(m)	(m)	(ppb)	(g/t)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)
197100	51.20	53.10	1.90	680		7.2	2350	45	34	24	258
197101	53.10	54.00	0.90	1680	1.61	42.4	1835	29	36	60	288
197102	54.00	55.35	1.35	1365	1.23	56.8	2690	24	56	64	260
197103	55.35	57.00	1.65	2340	2.33	11.0	5370	52	54	22	428
197104	57.00	59.74	2.74	<5		3.0	232	78	26	20	244
197105	59.74	61.40	1.66	1700	1.68	50.0	2640	34	202	34	656
Avg.	51.20	61.40	10.20	1080	1.08	23.3	2316	49	66	33	349
197109	69.50	71.00	1.50	250		1.8	1385	22	28	8	266
197112	81.40	83.90	2.50	160		1.0	332	35	14	6	124

# Table 7.0.7 Significant Intercepts - FWN97-06

### FWN97-07

Hole FWN 97-07 was the westernmost 1997 drill hole, collared 150 metres west of FWN97-06 on the V2 conductor, in an area with no soil geochemical anomalies and no surface epithermal float. It encountered a package of volcanic flows and fragmental volcanics which displayed alteration parallel to two strong faults (Figure 12). All quartz and dolomite stringering occur within the faults or alteration zone.

Pyrite mineralization occurs disseminated throughout the extremely altered zones, and within minor quartz stringers. Within the fault zones clay gouge and breccia appear to contain disseminated pyrite throughout. Fragments of quartz-feldspar porphyry within the fault breccia suggest the presence of a dyke which may have occupied the zone, but was reduced to rubble by faulting. The quartz feldspar porphyry fragments may also have been dragged into the zone from elsewhere.

Stringering throughout the hole appeared to be substantially less than previous holes along the V2 conductor. The reduction in stringering may be the result of the strong faulting which may have postdated the quartz veining event. The strong faults from 43.7 to 50.4 metres and 63.6 to 67.0 metres may both represent the V2 conductor although the 43.7-50.4 metre fault displayed a larger interval with strong clay gouge.

### <u>Table 7.0.8</u> Significant Intercepts - FWN97-07

Sample	From	То	Length	Au	Au	Ag	As	Cu	Pb	Sb	Zn
Number	(m)	(m)	(m)	(ppb)	(g/t)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)
197115	29.30	30.60	1.30	140		0.6	120	21	12	6	78
197127	65.40	67.00	1.60	165		0.8	1890	17	20	6	90
197128	78.10	78.90	0.80	25		13.4	32	4060	490	4	1610

### 8.0 DISCUSSION

The 1997 diamond drilling program focused on the Giver Zone, an east-west trending zone of faulting and sericite-clay alteration with local quartz stringer stockworks and chalcedonic breccias. It is marked on surface by sporadic arsenic, lead and zinc soil geochemistry and a strong VLF-EM conductor (V2). Eight holes tested the Giver Zone in 1994 and 1997, on seven sections along 900 metres of its strike length. Each of these holes produced wide altered intervals and samples exceeding 100 ppb Au; five returned intersections of 1-3 g/tonne Au, including hole FWN97-06 with 10.2 metres grading 1.08 g/tonne Au.

Two 1997 holes tested the Giver Splay, a weak, northwesterly-trending VLF-EM cross-structure to the Giver Zone. Epithermal veining which subcrops between the 1997 holes and the Giver Zone had yielded assays up 12.9 g/tonne Au in 1991. However, the drilling shows the Giver Splay to be a relatively narrow zone of faulting and mineralization; the best core sample assayed 2.02 g/tonne Au across a few tens of centimetres true width.

Quartz-feldspar porphyry dykes were encountered in several Giver Zone holes, locally altered and mineralized. They both follow fault zones and are locally fractured or brecciated by later faulting. They may be related to the Late Cretaceous Capoose Lake Batholith or may be feeders to Eocene Ootsa Lake Group volcanics. Their close spatial association with alteration and epithermal mineralization in drill holes and on surface suggests a genetic link as well, implying that Fawn epithermal mineralization is Cretaceous or Tertiary in age. Repeated fault, hydrothermal and igneous activity along the Giver Zone is suggested by the locally brecciated dyking and by silicified fault breccias cut by later veining and faulting.

The alteration along the Giver Zone (V2) is indicative of a strong and extensive epithermal mineralizing system. Despite widespread 'sniffs' of gold, no potentially economic gold mineralization has yet been discovered along the 900 metres of the V2 conductor which has been drill tested. However, less than 15% of the prospective strike length of the known conductors has been drilled. Conductor V2 itself has been defined over an additional 1,000 metre strike length and remains open to the east and west. Three other east-west VLF-EM conductors with associated soil geochemical anomalies were also defined by the 1991 survey, aggregating 4,600 metres of strike length and each remaining open in one or two directions; none of these has been tested by drilling. In addition, it appears likely that further parallel structures will be found to the south of the 1991 geophysical survey, including the Malaput Showing, an east-west silicified zone four kilometres south of the V2 conductor. Epithermal systems are characterised by strong vertical controls on mineralization. It may turn out that the drilled portion of V2 is too low (or too high) in the epithermal system and one of the other structures is exposed at a better level.

The Tommy prospect, located 17 kilometres south of the Fawn property, consists of epithermal quartz veins in Hazelton Group (Naglico Formation) quartz-phyric rhyolite tuffs. Teck Corp. has developed a reserve of 478,000 tonnes grading 8.7 g/tonne Au over a width of four metres at Tommy in a geological setting which is very similar to the Fawn property's. Not only does this bode well for the

possibility of discovering significant gold mineralization on the Fawn claims, but it suggests a possible rheological control on mineralization. At Tommy, the rhyolite host forms brittle fractures, along which the quartz veins are emplaced. In the Giver Zone, the less competent and esitic lapilli tuffs do not form discrete fractures, but rather wide zones of faulting, alteration and quartz stockworks, with more dispersed gold mineralization. It would be worthwhile extending the Fawn geophysical grid eastward over the Naglico Formation quartz-phyric rhyolite tuffs. If the known fault structures, and their associated hydrothermal fluids, extended over into the rhyolite, they may also be characterized by narrower but more focused, discrete quartz veins.

Respectfully submitted, EQUITY ENGINEERING LTD. CONTRACTING Henry J. Awmack, P.Eng.

Jim Lehtinen, P.Geo.

Vancouver, British Columbia October, 1997

### **APPENDIX A**

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APPENDIX B

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### STATEMENT OF EXPENDITURES

### STATEMENT OF EXPENDITURES

Fawn 1 Claim

March 24 - April 12, 1997

# PROFESSIONAL FEES AND WAGES

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Stewart Harris, Project Geologist		
30.188 days @ \$425/day	12,829.90	
Jim Lehtinen, P. Geo		
23.5 days @ \$425/day	9,987.50	
Mark Baknes, P.Geo.		
3.25 days @ \$425/day	1,381.25	
Warren Cole, Field Assistant		
20.5 days @ \$225/day	4,612.50	
Clerical		
0.25 hours @ \$25/hour	6.25	\$ 28,817.40
EXPENSES		
Accommodation	\$ 12,048.52	

	/ locolillinoudlell	¥ 12,010.02	
	Cat	15,263.48	
	Automobile Fuel	983.59	
	Bulk Fuel	3,830.87	
	Water Truck	14,595.00	
	Camp Supplies	162.53	
	Chemical Analyses	2,166.86	
	Courier	48.65	
	Freight	775.00	
	Airfare	4,361.70	
	Materials and Supplies	3,219.92	
	Meals	146.16	
•	Taxis and Airporters	48.77	
	Printing and Reproductions	102.47	
	Tolls and Airport Taxes	9.35	
	Automotive Expenses	53.86	
	Telephone Distance Charges	352.66	
	Truck Rental (Non-Equity)	3,795.39	\$ 61,964.78
SUB-COI	NTRACTS		
	Falcon Drilling Ltd:		
	Mob/Demob	9,165.00	

	Mob/Demob	9,165.00	
	Footage	38,375.50	
	Materials	467.09	
	Standby/Moves/Travel	10,290.00	
	Other Drilling Expenses	 1,960.00	\$ 60,257.59
EQUIPMENT R	ENTALS		
4x4	Equity Crewcab		
	20 days @ \$80/manday	\$ 1,600.00	
Gen	erator, 1kVA		
	5 days @ \$10/day	50.00	
Truc	k Radio		
	19 days @ \$5/day	95.00	

17 days @ \$15/day 255.00 Handheld Radios 12 days @ \$5/day60.00_	\$	2,060.00
REPORT (estimated)		
Drafting \$ 2,500.00		
Printing and Reproductions 900.00		
Time 6,600.00	\$	10,000.00
SUBTOTAL PROJECT SUPERVISION CHARGE	\$	163,099.77
12% on first \$100,000	¢	12,000.00
10% on remaining \$63,099.77	\$ \$	6,309.98
	<u></u>	0,503.50
SUBTOTAL	\$	181,409.75
GST		
7.0 % on subtotal (including project supervision charges)	\$	12,698.68
TOTAL	\$	194,108.43

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### APPENDIX C

### **DIAMOND DRILL LOGS**

### MINERALS AND ALTERATION TYPES

AS arsenopyrite CL chlorite EP epidote HE hematite MG magnetite PO pyrrhotite SI silica

BIbiotiteCPchalcopyriteGEgoethiteJAjarositeMNMn-oxidesPYpyriteSPsphalerite

CA calcite CY clay GL galena MC malachite MS sericite QZ quartz TT tetrahedrite

EUITY ENGINEERING LTD.
PROJECT

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4· 5·0 6	4	13						- Topilli tragments commonly pale green, also rare donk arey - Fragments up to 4 cm in LZMM - Ash matrix. Rove faths of foldspor crystals up to 1.5mm.						
 - 7 M 								- Strong Cridation along tractures Alteration - Variably correcte altel with trace homotife alteration thronglant. Pakty hematite pear base of interval.						
- - 9 - - - 10 -	ទា	24		•			11.3-	Henelile 19/tored Lapilli Tuff					54	
-    -	103	78				3	21.8	- Variably altered, dominantly marcon matrix with pale green-grey fragments. Fragments variable composition from ash full crystal full,						
- (4 	ß	59		•				form ash fuff crystal full, portypitic and amygdaloidal. - Minor 6 3mm dolow, & stringers at 40-55°TCA + along C.A. - Minor Cley olong weak slips - Tomes of moderate sericity alforetron from 170 to 18.2m + 11.8to 20.8						
- 16 - - 17 - 18	<u>8</u>	52				Star Partie		- 21.2 - Minon errefic chalcodarie quarte.					7	
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- 23 - 24 - 25.0	447	L				22.9- 33.0	- Weak calcile stringers 22mm - Strong calcite alteration Motthed Lapille Tuff. Variably coloured service Ittematic oldered fulf, vorging from pole						20
- 26 - 27 - 100 - 28	0 4	2		A A A			green-grey to varying brown-read - Fragments commonly of similar composition as motrix, but also. black - -Matrix of ash, & minor crystal fulff						· •
- 29 - -30 68 -31 _	8 31				<u> </u>	<u> </u>	Numerous stringers throughout I Dolomite, Calcile + County 22.9- QZ-CA Stringer @ 40° TCA - Strong forcture fol's Horackle						
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PAGE 4 OF 8 HOLE PROJECT FANN FWN 97-01 ASSAYS SAMPLES TOTAL SULPHIDE MINERALIZATION SAMPLE WIDTH NUMBER DESCRIPTION FROM то 2º - Chlorite contonnate Alt' Vole. - no Sx. 1.3 22.9 21-6 197008 21 Miner Calaite stringers 23mm Stringers near base & interval 22.9 12 - Strangly CA +DO strangeved at top of interna ( + Carbon? 24.2 1.3 197009 24-MS altid. went comben alt is 1.3 197010 24.2 25.5 near base 25.5 - weak azt DO stringers 14 with Trace PY. - MS + HE alt'n No SX 26.2 0.7 197011 ┼┼┽╴ 26.2 27.4 1.2 197012 - Nomerone 10 stringers 27.4 28.8 1.4 197013 L 5mm + miner Voin BX 28 29 30 31 - Bottom of MS alt'an -grading to strong HE alt'n . DO 32 stringers and creekle BX. 33-0 1.1 31.9 197014 1 32 39 55 36 37 38 39 40 -

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	19 - 78.9 -> 79.7 = az Bx - Continuentes of above, with 1-3 % fus. PY	Н	H	-	. <b>Le</b>									
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Strongly pyritical with PY surrounding		+						1			╉──
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-32.1 - 39.2 - Asper 197051	′ ₩	4	37.1	39.2	2.1	197051				<u> </u>	
18 Ales includes vuggy QZ-PY	<u> </u>	+			ļ			ļ		<u> </u>	_
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39 38.35 -7 75 TCA 39.2 to TCA	╶┟┟┼	+	1	<u> </u>	<u> </u>			ļ		<b> </b>	
- Strongly fractured MS/CY afferred	_ᠮᠯ╡	1	39.2	40.7	1.5	197052		<b> </b>			<b></b>
Strongly py extread with PY surrounding 1 1 mm ash flagments + also Py as fine 11mm stringers - The pyrite 4 altertim general which are be beige primeral which are be us the thimbonil - with difficulty) 54 Or some mixture of both - Banding appears to be 32 Pyrite alteration-Minor G2 shin -32.1 - 59.2 - HS per 197051 38 Ale includes viggy QZ-PY Strongers 38.7m - 35°TCA 39 38.35 - 75°TCA, 39.2m to TCA - Strongly Includes [CY aftered 40 Zohg Weak DO. Stringers			1		ļ						
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PAGE 8 OF 13 PROJECT HOLE FAWN Fm 97-04 SAMPLES ASSAYS TOTAL MINERALIZATION SAMPLE WIDTH DESCRIPTION NUMBER FROM то 55 3 3 St. a.t. 56-3 Sc-Grading from Homen DO+OZ 57.2 0.9 197057 Stringers at top to QZ t DO year base PY as dissen + stringers - AS dissen PY as dissen + strongers - AS dissen QZ = vaggy + up to scm near base. Sharp basal contect. DO: post dates 57 QZ. Drusy califies with QZ + laterDS. 57.2 57.6 0.4 197058 - Trace PY. c's dissen \_\_\_\_ - Drusy Caribies in Q2 Stringers 58 PY. L.g. providing verning + As < 1000 stringers. To ASGP Strong track fol northassoc. stringers A 57.6 58.7 1.10 19 7059 58.7 59.4 0.7 - Strong fracture tol'n (No Sx) 197060 59 59.40 59.8 0.4 - Very strong fracture to I'n. with assoc. PY + As strongening. 60 Dissem. PY Throughout - MS-7HE-> CY attered + fault 197061 1 59-8 61.5 1.7 197062 breccietar . Tr. PY TT ++61 61.5 -Fault zone, Fault Bx + 62.4 0.9 197063 clay eltered . Tr. Py ╉╫┼ 6Z 1 ТΠ 

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PAGE 10 OF 13 PROJECT FAV	ŝ								но	<sup>LE</sup> FN	10 97
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MINERALIZATION DESCRIPTION		סטררחוט	FROM	то	WIDTH	SAMPLE NUMBER					
6 (See Previous Page)	Ħ	E	59.8	61.5	1.7	197062					ļ
41 ····			61-5	67.4	0.9	197063				ļ	
						10-2					
62-Strugly Ault bx Wet PY Paralloling Blip - Dissem. PY in fragstgouge		╞	62.4	63.3	0.9	197064					
63 = Mamarone - 2 3mm QZ strongers with 14 t SP - Stockwork strongering - 64 Very showly fault broccoiled	<b>A</b>		63.3	65-0	1.7	197065					
14t SP Stockwork stringering -	14-										
ar Very shangly fault brocented	11	╞									<b>_</b>
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65 - Strongly DO stringered at 250°TCQ.		E	65-0	66.5	1.5	197066					
Storigly broken care runners		H		<b>.</b>							<u> </u>
- Dissen fy in broken core +		Ħ	66-5	67.4	0.9	197067					
6] remaint QZ stringers up to 1cm	H	E									
in finishy stringard core.		H									<b> </b>
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and the Ill and Allapy		$\square$	69.2	700	11	197068					
- This DO stringers, To. PY		Ħ	<u>(a)+ (</u>		1.6	17 1060					
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77-Pyrite Hole surrounding = sch az-Do Vein with minist PY	<b>—</b>		76.0	7.00		197069					
23 PY in wellneks 5-7% - Rock holacs peralleling QZ+DQ 24 strongers Single CA stronger P13/0 Gs dissens in wellrock			73.9	74-8	0.9	197070					
14 stringers Single CA stringer 17370	扭										
45 - Dissen PY in wellpock + 2 QZ	纠	H	74.8	15.9	111	19707/				<u> </u>	
storagens with PY+SP-Stronaly	曲		75.9	77.8	1.9	197072				<u> </u>	
storagers with PY + SP - Strongly 76 Practured + stronger ed	K	E									
- Assom Pitnumous DOt We Stringer	╞┿┿	$\vdash$				<u> </u>				<b> </b>	<u> </u>
73-Strong Protizetion of legilli Tell			778	79.3	1.5	197073					<b> </b>
- Competent are with QZ/DO 28 string ors Py. 3-5%						· · · ·					-
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29-Py. as discon Tr		F	79.3	81.5	2.2	197074					
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81	<u> </u>	-		┝━┦			39/	86.0	Entremely (br, He) fractured throughout.		1		+	Ħ	$\ddagger$	#1			15
				Ē	and		Æ		interval, with a common fracture		1		+	Ħ					
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-90						ЛJ	F		- Becoming coarse with lapelli	И				$\square$					HH.
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83 Quarte Feldspor Porphyry - No Visible Sx - Strongly tractured - CY+MS By alt's along fractures					19 200				
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DEPTH (M)	% CORE REC	% RQD			ПТНОГОВУ	STRUCTURE		GEOLOGICAL DESCRIPTION				RATIO			FRACTURE INTENSITY
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	MS-HEa			C		
18 7-719	ms allow	a Ach	T.Allavil	1; Tuff		
21.9 - 22.4	Fault in H	sh / Lape	11: T.A	n		· · · · · · · · · · · · · · · · · · ·
22.4-30.2	MS+HE a	Hered u	<u>apilli Tutt</u>			· · · · · · · · · · · · · · · · · · ·
20.7 - 41.35				······································		
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46.35 - 50-50				ing # Py	1201	
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46-35 - 50-50 Se-50 - 78,0 78,0 - 86,6 96,6 - 89,1 89,1 - 91,7		F. + A. Zame I S. Rish	shlip ///: + CZ + /- Lopille Te			

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## HOLE OF () PROJECT PAGE ۱ FWN 97-05 FAWN ALTERATION % CORE REC % RQD STRUCTURE INTENSITY **ПТНОLOGY** DEPTH (M) **GEOLOGICAL DESCRIPTION** HECA CY mS تحدثهم 0-3.66 ١ ٤ CASING TLAS 3-66-Logilli ·Andersite 3 17.7 med. dark oree Var 50 4 ·m a 54 0 -ix 37 \$ 0 117 50 Hir 78 15 ~ 7 83 0 CA 61 32 133 57 Do 8 r t a ne Ke. HE MS Mane 9 103 39 HE 200 10 120 1 634 ۰,۱ ιz 103 33 13 ĊÁ C) 14 15 98 39 16 CA. TA Londie 17.20 -Hen 17 at intervel are 18.70 top dine -m. 11E CY in bottom of inte cak )0 18 man 99 47 ₽¢ hase a 19 Q. fringers @ DO İπ 18.70-20 - 40 Z1.D gradina 2: Grainec 1 ash upper + Lower T4 ++ CHU 0 conte Diferance . Lo appers. 11

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							43.3 -	Saricite and Clay alteral forphysitic Andrich				И	$\pm$					
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t5.	145	39	╘─╋┈	-	17	$\leq$		- Fault @ 47.8 m - 25°TCH - Gauge + BX	Ħ	4	#	#	+	忄	Ħ			╞┼┼┨
				7				- Basal contact sharp @ 40°TCA	nt	4		<b>†</b> †	+-	は	╞╋			┇╡╡╄
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3	131	49			17	Г		Increased purchastion with	H	┦	H	H		$\mathbb{F}$	$\mathbb{H}$		$\mathbf{M}$	140
,	71	6			<b>F</b>			5 cm banded to liated DO-OZ	R	Ŧ		H		ſ	₽	+	$\mathbf{A}$	
				62/0	1-1	-		Vein of top d'interval	FJ	4	Π	F		K	Ħ		A	$\square$
	102	49			17			- Licm grey QZ stragers with	И	Ŧ	Ħ	Ħ	+	F .	Ħ		AÀ	
			┝─┼╴					PY - (Do-k grey =F.G. AS?)	H	4	##	Ħ	+	K			H/H	
6					11	-40		QZ = Chalcedarie	КЦ	#		Ħ	+	Ħ	Ħ		47	HZ.
	Az	39				t		WET CHAICENONIC	ГЦ	1	#	Ħ	+	11	╞╋		H/T	┇╧┽┚
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•		·		Q2 5	44	Įr.	<u>55.35</u>		怚	4	╞┼╴	#1	+-	┢╉	╞╋		#	┇┼┼┤
	107	41			11	'		- Zone comprised of unggy banded QZ+ DO veining with	И	1		╞	-			++	##	╊╪┿┥ ┨
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				<u> </u>	Ы	£		dominantly gray chate edonic	ť₽⊟	$\pm$		$\blacksquare$	+	¦∕上		╈		
•	⊢	$\vdash$	┝┈┼╸	- 777.	P	$\neq$		& white quarte	<b>K</b>	+							Шľ	129
	102	20		_	1	Y		- Stringer stock work and	M	Ⅎ	H		$\pm$	И	H		H	
S		<b>~</b>		A.S Q₹		ß		minor silicitied Zomes	K	Ŧ	$\prod$	H	F	H	H		47	
	$\vdash$		┝╌┾╸		1			- Minor vein breccia	FR	4	$\left[ \right]$	H	T	A	H		H	H
				0.1 k				- Mineralization = PY ± HS?	<b>M</b>	4		H	+	H	Ħ	++		
	105	20		a2/0	┞╴┧	Ħ		Secondary = Realgar, Tr. SP	<b>X</b>	井		Ħ		1/	₽₽	++-	μŤ	╞╦┼
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	95	18					55.35-	Strong Sericite Altered Porphyritic	ĽЬ	4			+		╡┨	11-	K† I	
3				1	╧		59.74	Andosite with interpedded Ash Tuff	К	$\downarrow$			+	Ґ	;∔	╪╪╴	1/1	1+++
								Pale green with a dark	Ħ	4				1		#		
	112	44							H	4					+			
						16		grainy appearance due to concentrations of PY as	$\mathbb{H}$	Z							⊬	
	⊢	-	╞╌┠╴	Lo. to SIB				alteration - Miner Tult (basal 1.0m)	4		1	П		Ł₽	H		4	<u>F</u> T
6									$\square$	+	11	H		<b>H</b>	Π			++
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48			A DOTRZ+	<u>፡፡</u> 	₽		50.0	51.2	<u> ·2</u>	19 7099					
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		Ľ-					- Pyrite with TE SP.	Z		Ħ		+	Ħ			LT Y	
4					4		- Amper-like mineral on fracture	H	Ľ	$\exists$			H	+		1/D	
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					1	}	- Disson PY in matrix . CY Gouge.	$\mathcal{P}$	T	H	H	Ŧ	Ħ	++	H	$\square$	
8	$\vdash$			1	1¢Z		- Strongly - Sk rich @	Ħ	<b>1</b>	Ħ		+	Ħ	$\ddagger$	<b>  </b>	<b> </b>	<b>1</b> ††
-					άz	<b></b>	61.8-62.2 ¥ 62.55 -62.5		4,	Ħ		+	‡		╞╪╪	111	┇┇┇
9					200	ł		H	$\mathbf{I}$		╢	╧		$\pm$		╉╉┼	┇┊╞
1	89	35			And a	62-5 -	Extremely Broken + Faulter!	H	4	+	$\pm$	+	╞		┢╁╋		
				6	r,	69.5	Clay-Serverte altered Ardesike	H	Ŧ	H	用	Ŧ	H	++	$H\overline{H}$	$\mathbf{H}$	+++
₩		- 1		00	<b> </b> 4_		- med- laft arean -grey.	T.	4	Ħ	Ħ	+	H	$\mp$	ĦŦ	$\Pi$	HA
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3						71. /	- Med-light grey-grown		-	Π	$\square$	Ŧ			+++	ŢĮĘĬ	1
	47	83				┇╡	- Med-light grey-grown . - Strongly DO stringe-ed.	#	+	#		1	Ħ	#	111	71	╀┼┼
4		:					-Py-ite as stringers (bonded) -Minor L som DZ with MY +AS. (Vuggy stringers)	山	+	#		1	Ħ	1	<u>†</u> ‡‡	#	╈
'							-Minor & som OF with MY + AS.	$\mathbb{H}$		┢┼		+	┢╋		┢┼┼		
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75	Γ		·		1-1-				+	Ħ	$\square$		H	++	╂╂╂	-MI	+++
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PAGE OF 10 HOLE FWID 97-06 PROJECT FAWN SAMPLES ASSAYS TOTAL SULPHIDE MINERALIZATION SAMPLE WIDTH DESCRIPTION NUMBER FROM то 59.74-Struck Sticked & Bx. PY. 1 59.74 61.4 Finally dissen PY. 7-10 % Strand Silichied & Br. PY. 1.66 197105 Finally dissen PY. 7-10 % 67 61.4 61.4-62.5 - Fault Bx with 62.5 1.1 197106 13 Bx Frags = Grey QZ with PY making of fault = finaly dission PY 64 Everall = 5% PY **n**t t 62.5-64.0- Footwall of 62.5 64.0 1.5 197107 fault and mineralized zone 66 67 68.0-69.5 68 Fault zone . Dissen PY 3% A 680 69.5 1.5 197108 with 7-10% in H.W. Fragments 69 69.5-71.0 = Strongly PV allored 169.5 71.0 1.5 197109 20 parallelling. NC stringiers - Minor L'Icm QZ-100 21 Stringers with Trace AS+SP \$ 71.0 72.4 1.4 197110 PT = 10-15% Overall 1 72 71.0-72.4- MSKPY 33 altid. Py altin as bands porolleling weak DO stringe 74 PY 3-5° 75 76 22 78 79 80 

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PAGE 6 PROJECT FAWIS OF 10 HOLE 97-07 SAMPLES ASSAYS TOTAL **MINERALIZATION** SAMPLE WIDTH DESCRIPTION NUMBER FROM то 40 Ľ. ┢╋┿┥ 41 47.10 43.7 1.6 47 42.10-43.7 Minon & 2000 black 197120 Cholesdonic Stringers Tr. PY 43.7 46.0 2.3 197121 43 43.7-460 - Strongly broken minop <u>Lemmy Vagy</u> QZ strongers <u>1</u> DO PY 3% 44 ∰ \*< 46.0 46 46.0 - 47.85 - Med-dkarry. Finally dissom + fradrame fill PY. 47.85 1.85 197122 
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-83	104	u					- Minin CA + EP + PY stringers throughout interval - String Magnetic											
- 94	136	51		X	2		78.1-78.8 = Fractures (Descripte with PY + CP fill (overall = 41% CP) - Basal contect = 50° TCP											
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-86	97	6			ŝ	90.22	Andesite Feldspor Porphyry Light grey phonos in dark grey											
-87 -87	 						-Phones up to 2mm, medacolaly packed - Very weak CH altin.											
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## 1997 DIAMOND DRILL CORE SAMPLES

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Sample	From	То	Length	Au	Au	Ag	As	Cu	Pb	Sb	Zn
Number	(m)	(m)	(m)	(ppb)	(g/t)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)
FWN97-01											
197001	3.05	4.55	1.50	30		1.0	122	3	86	4	880
197002	10.10	11.00	0.90	<5		0.8	14	23	60	8	570
197003	11.00	12.50	1.50	<5		<.2	10	<1	10	4	286
197004	16.00	17.00	1.00	<5		<.2	4	5	8	4	194
197005	17.00	18.20	1.20	5		1.0	6	57	16	<2	350
197006	18.20	19.83	1.63	<5		<.2	8	13	10	2	140
197007	19.83	21.60	1.77	<5		<.2	10	16	12	2	108
197008	21.60	22.90	1.30	<5		<.2	14	11	2	<2	78
197009	22.90	24.20	1.30	10		0.4	6	169	8	2	180
197010	24.20	25.50	1.30	<5		<.2	<2	1	2	<2	136
197011	25.50	26.20	0.70	<5		<.2	4	<1	4	2	246
197012	26.20	27.40	1.20	<5		<.2	12	<1	10	2	396
197013	27.40	28.80	1.40	<5		0.6	6	13	4	<2	420
197014	31.90	33.00	1.10	<5		0.4	14	44	10	14	142
197015	44.20	44.70	0.50	5		3.8	74	112	76	42	258
197016	47.60	49.60	2.00	<5		<.2	6	10	4	6	80
197017	56.00	57.10	1.10	<5		<.2	14	10	6	2	240
197018	57.10	58.80	1.70	<5		<.2	14	5	2	2	92
197019	58.80	60.15	1.35	365		2.8	928	27	60	20	440
197020	60.15	61.90	1.75	<5		0.4	32	8	34	6	178
197021	61.90	63.00	1.10	2000	2.02	6.0	2270	9	84	10	508
197022	63.00	64.00	1.00	10		<.2	26	2	12	8	112
FWN97-02											
197023	3.05	4.50	1.45	<5		1.0	36	23	28	4	372
197024	13.40	14.70	1.30	170		1.2	6	28	262	<2	484
197025	48.00	50.00	2.00	<5		<.2	16	50	16	12	122
197026	50.00	52.20	2.20	130		3.8	264	28	44	14	154
197027	52.20	53.20	1.00	<5		. <.2	22	9	8	<2	56
197028	56.90	58.50	1.60	20		0.4	166	5	14	6	142
197029	58.50	59.30	0.80	10		0.6	52	25	54	8	320
197030	60.70	62.50	1.80	<5		0.4	20	15	36	4	292
197031	62.50	64.00	1.50	20		2.6	12	18	60	<2	388
197032	64.00	64.90	0.90	15		1.8	20	81	56	6	398
FWN97-03											
197033	6.40	8.23	1.83	10		<.2	8	49	2	<2	96
197034	41.80	43.10	1.30	<5		<.2	88	44	6	2	88
197035	43.10	44.70	1.60	<5		0.2	44	46	2	8	88
197036	44.70	46.20	1.50	<5		<.2	2	12	<2	2	80
197037	46.20	47.30	1.10	<5		<.2	12	6	2	<2	94
197038	51.70	54.00	2.30	5		0.4	56	25	4	2	100
197039	63.15	65.53	2.38	10		0.2	306	23	8	2	138
197040	65.53	67.60	2.07	<5	***	0.4	28	19	10	4	122
197041	67.60	70.41	2.81	60		0.6	254	7	38	2	246
197042	70.41	74.10	3.69	575		2.2	1750	11	68	6	348
197043	76.20	77.00	0.80	25		0.6	206	2	18	2	246
197044	77.00	78.90	1.90	345		1.4	1530	12	24	8	186
197045	78.90	81.50	2.60	345		1.8	1420	14	36	8	224

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Sample	From	To (mr)	Length	Au (mmh)	Au	Ag	As	Cu	Pb (nnm)	Sb (nnm)	Zn (nnm)
Number	(m)	(m)	(m)	(ppb)	(g/t)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)
FWN97-04	0.70	11 20	1 60	<5		0.6	24	42	6	8	114
197046	9.70	11.30	1.60			0.6	24 16	42 37	38	6	206
197047	11.30	12.90	1.60	<5						6	200
197048	12.90	14.30	1.40	<5		0.2	18	31	2 2	12	
197049	14.30	15.80	1.50	130		<.2	14	46			120
197050	33.40	35.10	1.70	80		2.4	508	16	14	16	122
197051	37.10	39.20	2.10	<5		3.2	214	17	14	12	128
197052	39.20	40.70	1.50	<5		0.8	24	27	2	6	46
197053	46.50	47.50	1.00	<5		0.6	58	28	6	6	94
197054	48.50	49.00	0.50	30		1.4	172	11	12	12	96 64
197055	49.50	50.80	1.30	<5		0.4	36	28	6	8	64 450
197056	50.80	52.10	1.30	70		2.8	212	28	16	14	156
197057	56.30	57.20	0.90	125		1.2	282	10	16	8	70
197058	57.20	57.60	0.40	<5		0.2	36	8	10	6	104
197059	57.60	58.70	1.10	130		3.2	370	27	18	12	176
197060	58.70	59.40	0.70	<5		<.2	18	4	2	2	84
197061	59.40	59.80	0.40	580		6.6	3690	14	50	14	266
197062	59.80	61.50	1.70	<5		<.2	20	7	<2	2	82
197063	61.50	62.40	0.90	10		0.2	176	7	10	4	142
197064	62.40	63.30	0.90	30		0.6	500	12	12	6	224
197065	63.30	65.00	1.70	140		1.8	622	11	30	8	184
197066	65.00	66.50	1.50	<5		0.2	16	8	6	<2	100
197067	66.50	67.40	0.90	120		1.2	1200	17	18	10	128
197068	69.20	70.80	1.60	<5		<.2	26	13	<2	<2	98
197069	72.20	73.00	0.80	30		0.6	150	7	20	4	618
197070	73.90	74.80	0.90	50		0.4	220	29	8	8	142
197071	74.80	75.90	1.10	275		2.2	638	17	24	8	284
197072	75.90	77.80	1.90	10		1.0	54	38	24	6	908
197073	77.80	79.30	1.50	50		1.0	152	34	6	10	86
197074	79.30	81.50	2.20	30		0.6	98	34	<2	10	102
197080	83.82	85.04	1.22	<5		<.2	8	<1	6	<2	8
197081	86.00	87.30	1.30	10		2.4	32	47	8	8	76
FWN97-05							_				
197075	18.00	20.00	2.00	<5		0.2	26	103	4	24	80
197076	20.00	21.00	1.00	<5		<.2	52	18	14	6	140
197077	21.00	21.90	0.90	<5		<.2	32	8	30	12	124
197078	21.90	22.40	0.50	10		0.2	246	42	12	14	138
197079	22.40	24.20	1.80	<5		<.2	80	39	2	6	72
197082	50.00	50.50	0.50	<5		0.2	70	18	10	6	112
197083	50.50	52.10	1.60	95		1.8	752	2	12	8	170
197084	52.10	53.03	0.93	60		1.4	142	1	8	<2	12
197085	53.03	54.50	1.47	25		3.8	46	10	10	4	18
197086	54.50	56.00	1.50	10		3.2	20	10	10	4	14
197087	56.00	57.50	1.50	5		0.8	36	5	12	2	14
197088	57.50	59.00	1.50	10		1.4	34	4	8	<2	12
197089	59.00	60.50	1.50	10		0.8	30	1	10	2	8
197090	60.50	62.80	2.30	55		0.6	120	1	14	<2	18
197091	62.80	65.90	3.10	20		<.2	28	<1	2		2
197092	86.60	87.90	1.30	575		4.4	1295	8	10		74
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Sample	From	То	Length	Au	Au	Ag	As	Cu	Pb	Sb	Zn
Number	(m)	(m)	(m)	(ppb)	(g/t)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)	(ppm)
FWN97-05											
197093	87.90	89.10	1.20	210		3.0	892	12	12	8	106
197094	89.10	90.60	1.50	1620	1.58	0.2	16	31	62	2	312
197095	90.60	91.70	1.10	300		0.6	62	47	56	6	368
197096	91.70	93.30	1.60	<5		0.8	114	35	6	8	38
197097	94.60	96.31	1.71	<5		0.2	114	62	2	16	58
FWN97-06											
197098	47.30	49.00	1.70	30		0.4	282	30	30	8	370
197111	49.00	50.00	1.00	15		<.2	32	13	20	2	344
197099	50.00	51.20	1.20	30		0.8	240	37	40	12	376
197100	51.20	53.10	1.90	680		7.2	2350	45	34	24	258
197101	53.10	54.00	0.90	1680	1.61	42.4	1835	29	36	60	288
197102	54.00	55.35	1.35	1365	1.23	56.8	2690	24	56	64	260
197103	55.35	57.00	1.65	2340	2.33	11.0	5370	52	54	22	428
197104	57.00	59.74	2.74	<5		3.0	232	78	26	20	244
197105	59.74	61.40	1.66	1700	1.68	50.0	2640	34	202	34	656
197106	61.40	62.50	1.10	65		3.0	246	76	84	22	460
197107	62.50	64.00	1.50	<5		0.2	16	31	64	2	314
197108	68.00	69.50	1.50	10		0.6	66	48	54	4	380
197109	69.50	71.00	1.50	250		1.8	1385	22	28	8	266
197110	71.00	72.40	1.40	<5		<.2	14	26	2	2	116
197112	81.40	83.90	2.50	160		1.0	332	35	14	6	124
197113	85.60	86.20	0.60	25		1.0	32	27	10	2	96
197114	88.00	89.90	1.90	<5		<.2	6	64	6	<2	104
FWN97-07						~ ~	400	•	40	•	70
197115	29.30	30.60	1.30	140		0.6	120	21	12	6	78
197116	30.60	32.30	1.70	<5		<.2	12	59	2	<2	92
197117	32.30	33.40	1.10	<5		<.2	32	33	6	4	78
197118	33.40	35.35	1.95	<5		<.2	30	33	6	6	314
197119	35.35	37.50	2.15	<5		0.2	32	68	6	14	98
197120	42.10	43.70	1.60	<5		<.2	10	60	8	14	66
197121	43.70	46.00	2.30	30		0.4	188	35	12	14	98
197122	46.00	47.85	1.85	10		0.2	20	54	8	16	60
197123	47.85	48.70	0.85	<5		<.2	22	22	8	4	38
197124	48.70	49.68	0.98	30		<.2	12	9	6	<2	30
197125	49.68	50.40	0.72	30		0.6	94	76	12	16	84
197126	63.60	65.40	1.80	<5 465		<.2	8 1900	26	8	<2	68
197127	65.40	67.00	1.60	165		0.8	1890	17	20	6	90 1610
197128	78.10	78.90	0.80	25		13.4	32	4060	490	4	1610
197129	87.17	88.20	1.03	<5		<.2	16	50	8	<2	90

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**APPENDIX D** 

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## **CERTIFICATES OF ANALYSIS**

\_\_\_\_ Equity Engineering Ltd. \_



## Chemex Labs Ltd.

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Analytical Chemists \* Geochemists \* Registered Assayers 212 Brooksbank Ave., North Vancouver British Columbia, Canada V7J 2C1 PHONE: 604-984-0221 FAX: 604-984-0218

A9720423

✓Fo: EQUITY ENGINEERING LTD.

 $\square$ 

207 - 675 W. HASTINGS ST. VANCOUVER, BC V6B 1N2

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Comments: ATTN:S.HARRIS/J.LEHTINEN

DETECTION LIMIT 5 0.07 0.2 0.01 2 10 0.5 2 0.01 0.5 1 1 0.01 10 0.01 10 0.01 10 0.01 10 0.01 10 0.01 10 0.01 10 0.01 10 0.01 10 0.01 10 0.01 10 0.5 10 10 0.5 10 10 10 10 10 10 10 10 10 10	UPPER LIMIT 10000 10000 10000 10000 10000 10000 10000 10000 10000 10000 10000 10000 10000 10000 10000 15.00 10000 15.00 10000
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10 0.5 2 0.01 0.5 1 1 1 0.01 10 10 0.01 10 0.01 5 1	$10000 \\ 100.0 \\ 10000 \\ 15.00 \\ 10000 \\ 10000 \\ 10000 \\ 15.00 \\ 10000 \\ 10000 \\ 10000 \\ 10000 \\ 10.00 \\ 15.00 $
0.5 2 0.01 0.5 1 1 0.01 10 1 0.01 10 0.01 5 1	100.0 10000 15.00 10000 10000 10000 15.00 10000 10.000 10.00 15.00
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(EIA ) - EQUITY ENGINEERING LTD.

CERTIFICATE

Project: WKM 97-06 P.O. # :

Samples submitted to our lab in Vancouver, BC. This report was printed on 21-AUG-97.

		SAMPLE PREPARATION							
Hemex Code	NUMBER SAMPLES	DESCRIPTION							
205 226 3202 229	129 129 129 129	Geochem ring to approx 150 mesh 0-3 Kg crush and split Rock – save entire reject ICP – AQ Digestion charge							

The 32 element ICP package is suitable for trace metals in soil and rock samples. Elements for which the nitric-aqua regia digestion is possibly incomplete are: Al, Ba, Be, Ca, Cr, Ga, K, La, Mg, Na, Sr, Ti, Tl, W. A9720423

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#### Chemex Labs Ltd.

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Analytical Chemists \* Geochemists \* Registered Assayers

212 Brooksbank Ave., North Vancouver British Columbia, Canada V7J 2C1 PHONE: 604-984-0221 FAX: 604-984-0218 /ro: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST. VANCOUVER, BC V6B 1N2 Page Joer :1-A Total Pages :4 Certificate Date: 21-AUG-97 Invoice No. : 19720423 P.O. Number : Account : EIA

 $\square$ 

Project : WKM 97-06 Comments: ATTN:S.HARRIS/J.LEHTINEN

											YSIS	4	9720	423					
SAMPLE	PREP CODE	Au ppb Au FA FA+AA g/t		Al %	As ppm	Ba ppm	Be ppm	Bi ppm	Ca %	Cd ppm	Co ppm	Cr ppm	Cu ppm	Fe %	Ga ppm	Hg mgq	K %	La ppm	Mg %
197001	205 226	30	1.0	0.60	122	20	0.5	2	0.77	3,5	22	24	3	5,21	< 10	1	0.28	< 10	0.38
197002	205 226	< 5	0.8	0.75	14	30	0.5	< 2	0.66	1.0	19	16	23	4.94	< 10	1	0.28	10	0.40
197003	205 226	< 5	< 0.2	0.99	10	30	0.5	< 2	0.86	< 0.5	19	12	< 1	5.79	< 10	< 1	0.31	10	0.48
197004 197005	205 226 205 226	<pre>&lt; 5</pre>	< 0.2 1.0	0.74 0.46	4	30 20	0.5 0.5	< 2 < 2	1.83 2.99	< 0.5 0.5	17 19	12 9	5 57	4.37 4.53	< 10 < 10	$\langle 1 \\ \langle 1 \rangle$	0.26 0.20	< 10 < 10	0.75
107006																			-
197006 197007	205 226 205 226	<pre> &lt; 5 &lt; 5</pre>	< 0.2 < 0.2	0.82 0.67	8 10	30 30	1.0	< 2 < 2	1.89	< 0.5 < 0.5	17	11	13	4.55 4.59	< 10	$\langle 1 \rangle$	0.27	10	0.68
197008	205 226	< 5	< 0.2	2.22	14	60	1.0 0.5	< 2	4.33	< 0.5	20 19	12 15	16 11	4.59	< 10 < 10	<1 <1	0.20 0.18	10 10	0.87
197009	205 226		0.4	0.86	6	30	0.5	ζ2	3.31	0.5	21	15	169	5.06	< 10	$\langle 1$	0.20	< 10	1.00
197010	205 226		< 0.2	0.93	< 2	30	0.5	< 2	1.02	< 0.5	18	7	1	5.17	< 10	< ī	0.23	< 10	0.51
197011	205 226	< 5	< 0.2	0.94	4	20	0.5	< 2	1.04	0.5	22	10	< 1	6.82	< 10	< 1	0.21	10	0.62
197012	205 226	< 5	< 0.2	1.01	12	20	0.5	(2	0.72	0.5	21	7	< 1	4.73	< 10	<1	0.23	< 10	0.36
197013	205 226	< 5	0.6	0.91	6	40	0.5	< 2	2.22	0.5	22	10	13	4.30	< 10	1	0.23	< 10	0.82
197014	205 226		0.4	0.91	14	30	0.5	< 2	3.61	< 0.5	19	11	44	4.70	< 10	< 1	0.20	< 10	1.17
197015	205 226	5	3.8	0.74	74	30	0.5	< 2	0.52	1.5	19	22	112	2.45	< 10	1	0.40	< 10	0.25
197016	205 226	< 5	< 0.2	0.82	6	20	0.5	< 2	3.00	< 0.5	16	9	10	4.41	< 10	< 1	0.20	10	0.83
197017	205 226	< 5	< 0.2	0.81	14	30	0.5	< 2	1.16	0.5	16	8	10	4.07	< 10	< 1	0.26	10	0.44
197018	205 226	< 5	< 0.2	0.72	14	20	0.5	< 2	0.99	< 0.5	21	B	5	5.53	< 10	< 1	0.25	10	0.42
197019	205 226	365	2.8	0.55	928	30	0.5	< 2	1.15	3.0	29	14	27	5.09	< 10	< 1	0.32	< 10	0.30
197020	205 226	< 5	0.4	0.79	32	30	1.0	< 2	0.59	0.5	27	7	8	5.92	< 10	< 1	0.32	10	0.31
197021	205 226	2000 2.02	-	0.77	2270	40	1.0	< 2	2.03	3.5	25	20	9	5.70	< 10	1	0.39	< 10	0.50
197022	205 226		< 0.2	0.87	26	20	1.0	< 2	1.98	< 0.5	20	9	2	6.03	< 10	< 1	0.22	< 10	0.72
197023	205 226	< 5 170	1.0	0.99	36	30	1.0	< 2	0.58	2.5	27	10	23	6.10	< 10	< 1	0.20	10	0.12
197024 197025	205 226 205 226		1.2 < 0.2	1.70 1.87	6 16	30 30	< 0.5 1.5	< 2 < 2	3.83 2.64	2.0 0.5	20 22	15 15	28 50	4.85 5.77	< 10 < 10	1 < 1	0.23 0.29	< 10 < 10	1.29 0.84
107026	205 226	110	3.8	0 70	264		1.5	( )	0 70	0.5	20	10	20	E 01	/ 10		0.36	. 10	0.16
197026 197027	205 226	130	< 0.2	0.79 1.00	264 22	30 30	1.5 1.5	< 2 < 2	0.72 1.04	0.5 < 0.5	28 23	10 10	28 9	5.21 6.16	< 10 < 10	< 1 1	0.36 0.26	< 10 10	0.16
197028	205 226		0.4	0.69	166	40	1.0	$\langle \hat{2} \rangle$	2.00	< 0.5	29	14	5	5.35	< 10	1	0.32	< 10	0.55
197029	205 226	10	0.6	0.64	52	40	0.5	< 2	0.43	1.5	21	11	25	4.05	< 10	< 1	0.36	10	0.12
197030	205 226		0.4	0.68	20	30	0.5	< 2	0.51	1.0	23	12	15	5.01	< 10	< 1	0.31	10	0.34
197031	205 226	20	2.6	0.65	12	30	0.5	< 2	1.52	1.5	30	15	18	6.07	< 10	< 1	0.29	< 10	0.53
197032	205 226	15	1.8	0.76	20	30	0.5	< 2	2.81	2.5	18	13	81	4.37	< 10	< ī	0.29	< 10	0.75
197033	205 226	10	< 0.2	2.46	8	40	< 0.5	< 2	4.53	< 0.5	15	21	49	4.31	< 10	< 1	0.18	< 10	0.97
197034	205 226	< 5	< 0.2	0.87	88	30	1.0	< 2	1.82	< 0.5	19	12	44	4.37	< 10	< 1	0.36	10	0.60
197035	205 226	< 5	0.2	0.85	44	40	0.5	< 2	2.82	< 0.5	18	17	46	4.50	< 10	< 1	0.31	< 10	0.86
197036	205 226	< 5	< 0.2	0.87	2	160	0.5	< 2	3.42	< 0.5	17	12	12	4.66	< 10	< 1	0.17	< 10	1.14
197037	205 226	< 5	< 0.2	0,91	12	30	0.5	< 2	2.12	< 0.5	21	13	6	5.32	< 10	< 1	0.14	< 10	0.89
197038	205 226	5	0.4	0.86	56	110	0.5	< 2		< 0.5	17	13	25	4.31	< 10	< 1	0.32	< 10	1.16
197039	205 226	10	0.2	1.03	306	40	1.0	< 2	1.37	< 0.5	16	14	23	4.48	< 10	1	0.39	10	0.44
197040	205 226	< 5	0.4	1.19	28	30	1.5	< 2	0.95	< 0.5	14	12	19	4.48	< 10	< 1	0.33	10	0.37
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CERTIFICATION:\_\_\_\_

Hart Brahler



# Chemex Labs Ltd.

Analytical Chemists \* Geochemists \* Registered Assayers

212 Brooksbank Ave.,North VancouverBritish Columbia, CanadaV7J 2C1PHONE: 604-984-0221FAX: 604-984-0218

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST. VANCOUVER, BC V6B 1N2

Project : WKM 97-06 Comments: ATTN:S.HARRIS/J.LEHTINEN

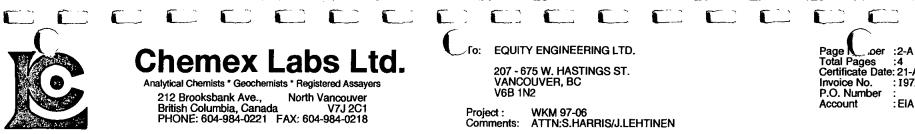
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Page Concert :1-B Total Pages :4 Certificate Date: 21-AUG-97 Invoice No. :19720423 P.O. Number : Account :EIA

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**CERTIFICATE OF ANALYSIS** A9720423 PREP Ni P Pb Sb Sc Ti **T1** U V W Zn Mn Мо Na Sr SAMPLE CODE ppm ppm 8 ppm ppm ppm ppm 8 ppm ppm ppm ppm ppm ppm ppm 205 226 < 1 < 0.01 1130 15 < 0.01 880 197001 6010 86 < 10 < 10 30 < 10 3 4 4 197002 205 226 6880 < 1 < 0.01 1500 60 8 14 < 0.01 < 10 32 < 10 570 2 < 10 4 197003 205 226 4430 < 1 0.01 1420 10 4 8 15 < 0.01 < 10 < 10 61 10 286 1 197004 205 226 2060 < 1 0.01 3 1090 8 4 5 33 < 0.01 < 10 < 10 49 < 10 194 197005 205 226 3360 < 1 < 0.01 2 1300 16 < 2 4 38 < 0.01 < 10 < 10 35 < 10 350 205 226 2390 < 1 < 0.01 1320 10 2 7 31 < 0.01 < 10 < 10 63 < 10 140 197006 3 205 226 1900 < 1 < 0.01 1370 12 2 8 75 < 0.01 < 10 < 10 60 < 10 108 197007 3 197008 205 226 1395 < 1 0.02 1280 2 < 2 6 77 < 0.01 < 10 < 10 62 < 10 78 3 < 10 180 197009 205 226 3090 < 1 < 0.01 4 1280 8 2 8 40 < 0.01 < 10 < 10 72 197010 205 226 5110 < 1 < 0.01 < 1 1310 2 < 2 7 15 < 0.01 < 10 < 10 66 < 10 136 4510 < 1 < 0.01 1380 16 < 0.01 < 10 94 < 10 246 205 226 4 2 8 < 10 197011 3 205 226 3790 < 1 < 0.01 1450 2 15 < 0.01 < 10 < 10 69 < 10 396 197012 10 11 2 < 10 205 226 3900 20 < 0.01< 10 < 10 54 420 197013 < 1 < 0.01 1210 < 2 6 3 4 < 10 142 205 226 2780 < 1 < 0.01 1200 10 7 59 < 0.01 < 10 < 10 83 197014 4 14 1965 < 1 < 0.01 910 42 12 < 0.01 < 10 < 10 < 10 258 197015 205 226 3 76 6 38 205 226 1635 < 1 < 0.01 1380 4 6 6 36 < 0.01 < 10 < 10 65 10 80 197016 3 205 226 3700 < 1 < 0.01 1540 6 2 17 < 0.01 < 10 < 10 42 < 10 240 197017 6 1 205 226 5790 12 < 0.01 < 10 68 < 10 92 197018 < 1 < 0.01 1410 2 2 8 < 10 1 205 226 4800 1190 20 6 14 < 0.01 < 10 < 10 31 < 10 440 197019 1 < 0.01 60 3 205 226 4990 < 1 < 0.01 1400 34 6 9 11 < 0.01 < 10 < 10 54 < 10 178 197020 2 197021 205 226 4630 1 < 0.01 3 1330 84 10 7 20 < 0.01 < 10 < 10 34 < 10 508 205 226 4590 1370 12 34 < 0.01 < 10 < 10 107 10 112 197022 < 1 < 0.01 < 1 8 10 < 10 205 226 3740 < 1 < 0.01 1420 28 4 10 20 < 0.01< 10 89 < 10 372 197023 3 205 226 3 62 < 0.01 < 10 < 10 38 < 10 484 5230 < 1 0.01 1430 262 < 2 197024 1 205 226 1885 < 1 < 0.01 1210 16 12 6 33 < 0.01 < 10 < 10 67 < 10 122 197025 3 197026 205 226 2180 < 1 < 0.01 4 1260 44 14 4 12 < 0.01 < 10 < 10 21 < 10 154 197027 205 226 >10000 < 1 < 0.01 5 1230 8 < 2 8 14 < 0.01 < 10 < 10 56 < 10 56 205 226 4050 < 1 < 0.01 1200 14 10 17 < 0.01 < 10 < 10 30 < 10 142 197028 -5 6 197029 205 226 1595 1 < 0.01 1380 54 8 5 11 < 0.01 < 10 < 10 23 < 10 320 4 197030 205 226 3290 < 1 < 0.01 1380 36 4 7 14 < 0.01 < 10 < 10 36 < 10 292 1 205 226 5340 0.01 1390 60 < 2 5 25 < 0.01 < 10 < 10 36 < 10 388 197031 5 2 197032 205 226 8250 1 0.01 4 1170 56 6 3 40 < 0.01 < 10 < 10 35 < 10 398 205 226 2080 2 2 2 95 < 0.01 < 10 < 10 49 < 10 96 197033 1135 1 0.08 2 ۲ < 10 88 197034 205 226 2570 < 1 < 0.01 1480 6 2 4 55 < 0.01 < 10 < 10 39 1 < 10 197035 205 226 2110 < 1 < 0.01 1320 2 8 4 54 < 0.01 < 10 < 10 40 88 1 205 226 5 66 < 0.01 < 10 < 10 < 10 80 197036 1385 < 1 < 0.01 1 1420 < 2 2 64 197037 205 226 1900 < 1 < 0.01 1 1430 2 < 2 6 36 < 0.01 < 10 < 10 75 < 10 94 205 226 1745 1140 6 71 < 0.01 < 10 < 10 49 < 10 100 197038 < 1 < 0.01 2 4 2 138 205 226 4280 1690 8 2 6 40 < 0.01 < 10 < 10 35 < 10 197039 < 1 < 0.01 1 205 226 10 5 34 < 0.01 < 10 < 10 43 < 10 122 197040 4590 < 1 < 0.01 < 1 1670 4 Starry & Starry



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## Chemex Labs Ltd.

Analytical Chemists \* Geochemists \* Registered Assavers 212 Brooksbank Ave., North Vancouver British Columbia, Canada V7J 2C1 PHONE: 604-984-0221 FAX: 604-984-0218

Fo: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST. VANCOUVER, BC V68 1N2

Project : WKM 97-06 Comments: ATTN:S.HARRIS/J.LEHTINEN

**CERTIFICATE OF ANALYSIS** 

Page Loer :2-A Total Pages :4 Certificate Date: 21-AUG-97 Invoice No. : 19720423 P.O. Number : EIA Account

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0.16

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Au ppb Au FA **A**1 Bi Ag As Ba Be Ca Cđ Co Cr Cu Fe Ga Ħg % FA+AA q/t ppm ppm ppm ppm ppm 8 ррл 8 ppm ppm ppm ppm ppm 60 -----0.6 1.09 254 70 < 2 1.0 0.60 1.0 16 34 7 3.14 < 10 575 -----2.2 0.40 1750 20 0.5 < 2 1.18 1.5 18 42 11 2.63 < 10 < 1 25 -----0.6 0.87 206 30 1.5 < 2 1.13 0.5 32 12 2 5.44 < 10 < 1 345 -----0.56 1530 1.4 20 1.0 < 2 0.62 0.5 26 24 12 3.76 < 10 < 1 345 -----1.8 1.02 1420 30 1.0 < 2 0,83 2.0 22 24 14 3.54 < 10 < 1 < 5 -----0.6 0.72 24 30 0.5 < 2 0,65 < 0.5 27 10 42 5.55 < 10 < 1 < 5 -----1.16 16 0.6 60 0.5 < 2 0.52 2.0 22 14 37 5.32 < 10 < 1 < 5 -----0.2 0,86 18 50 0.5 < 2 0.71 < 0.5 24 12 31 5.22 < 10 < 1 130 -----< 0.2 1.14 14 90 1.88 < 0.5 0.5 < 2 18 15 46 5.00 < 10 80 -----2.4 0.97 508 30 1.27 < 0.5 2.0 (2 27 8 16 5,67 < 10 < 5 -----1.32 214 40 1.5 < 2 0.51 3.2 < 0.5 24 16 17 4.82 < 10 < 5 -----0.8 0.99 24 100 1.5 < 2 0.69 < 0.5 26 10 27 5.46 < 10 < 5 -----0.6 1.36 58 30 1.5 < 2 0.80 < 0.5 21 14 28 3.89 < 10 30 -----1.17 172 20 1.4 2.0 < 2 0.64 < 0.5 29 7 11 5.99 < 10 < 5 -----0.4 36 1.29 30 2.0 < 2 0.85 < 0.5 22 8 28 5.36 < 10 70 -----2.8 0.75 212 1.5 0.62 < 0.5 7 28 5.36 10 < 2 26 < 10 125 -----1.2 0.94 282 40 1.0 < 2 0.61 21 10 < 0.5 17 4.32 < 10 < 5 -----0.2 0.70 36 110 1.0 < 2 0.63 < 0.5 18 7 8 5.42 < 10 130 -----370 3.2 1.14 40 1.5 < 2 0.50 0.5 16 24 27 3.78 < 10 < 5 -----< 0.2 0.89 18 30 1.5 0.99 < 0.5 16 7 < 2 4 4.67 < 10 580 -----6.6 1.40 3690 30 2.0 < 2 0.61 < 0.5 29 14 14 4.26 < 10 < 5 ~----< 0.2 0.92 20 100 15 1.0 < 2 1.33 < 0.5 7 7 3.94 < 10

< 1 0.40 10 0.43 < 10 1 < 1 0.70 < 10 < 1 0.27 < 10 0.74 10 1 < 1 0.46 < 10 0.52 1 < 10 0.44 < 1 < 10 < 10 1 0.52 < 1 0.33 < 10 < 1 0.53 10 1 0.30 10 < 1 0.59 10 < 1 0.32 10 1.27 176 0.2 40 1.5 < 2 0.80 < 0.5 15 11 3,86 < 10 < 1 0,52 10 - 7 0.6 0.93 500 30 1.07 15 4.05 1.5 < 2 0.5 R 12 < 10 < 1 0.44 10 622 1.8 1.02 80 1.0 < 2 0.94 < 0.5 17 10 11 4.46 0.43 10 < 10 < 1 0.85 16 40 0.5 < 2 1.29 < 0.5 20 9 8 4.54 < 10 < 1 0.36 10 1.10 1200 40 1.0 < 2 0.81 < 0.5 24 17 17 4.54 < 10 1 0.53 10 0.83 26 30 0.5 < 2 1.26 < 0.5 20 10 13 4.39 < 1 10 < 10 0.29

205 226 < 5 -----0.2 0.51 205 226 120 -----1.2 0.26 205 226 < 5 -----< 0.2 0.50 30 -----205 226 0.6 1.03 150 60 1.5 < 2 2,91 2.0 21 16 7 5.35 < 10 < 1 0.55 < 10 0.82 205 226 50 -----0.69 220 80 1.79 0.4 1.0 < 2 < 0.5 21 9 29 4.82 0.34 < 10 < 1 10 0.60 205 226 275 -----0.88 1.0 < 2 1.94 < 1 2.2 638 60 0.5 21 19 17 4.80 < 10 0.39 < 10 0.60 10 -----205 226 1.0 0.60 54 80 0.5 < 2 1.74 37 3.0 10 38 5.52 < 10 < 1 0.37 < 10 0.59 205 226 50 -----1.07 152 100 1.0 1.0 < 2 2.27 < 0.5 26 12 34 5.76 < 10 1 0.45 < 10 0.95 205 226 30 -----0.6 0.59 98 50 1.0 < 2 4.00 < 0.5 30 16 34 5.08 < 10 1 0.35 < 10 1.33 205 226 < 5 -----0.2 1.44 26 70 1.0 < 2 0.47 < 0.5 16 16 103 3.33 < 10 < 1 0.31 10 0.21 205 226 < 5 -----< 0.2 0.87 52 40 0.5 < 2 0.45 < 0.5 18 14 3,68 < 10 18 < 10 < 1 0.45 0.20 205 226 < 5 -----32 < 0.2 0.62 30 0.5 < 2 0,34 < 0.5 13 15 8 2.41 < 10 < 1 0.26 < 10 0.18 205 226 10 -----0.2 0.66 246 30 0.5 < 2 0.45 < 0.5 20 12 42 3.92 < 10 < 1 0.32 < 10 0.19 < 0.2 205 226 < 5 -----1.41 80 60 1.5 < 2 1,72 < 0.5 18 12 39 4.73 0.43 10 0.66 < 10 < 1 < 5 -----205 226 < 0.2 0.48 8 < 10 0.5 < 2 0.27 < 0.5 < 1 26 < 1 0.38 < 10 < 1 0.27 < 10 0.04



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#### **Chemex Labs Ltd.**

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Analytical Chemists \* Geochemists \* Registered Assayers

212 Brooksbank Ave., North Vancouver British Columbia, Canada V7J 2C1 PHONE: 604-984-0221 FAX: 604-984-0218 / To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST. VANCOUVER, BC V6B 1N2

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Project : WKM 97-06 Comments: ATTN:S.HARRIS/J.LEHTINEN Page ber :2-B Total Pages :4 Certificate Date: 21-AUG-97 Invoice No. : 19720423 P.O. Number : Account : EIA

A9720423

P Pb Sb Sc Sr Ti T1 U V W

SAMPLE	PREP CODE	Mn ppm	Mo ppm	Na %	Ni ppm	P ppm	Pb ppm	Sb ppm	Sc ppm	Sr ppm	Ti %	Tl PPm	U ppm	v ppm	w ppm	Zn ppm	
197041	205 226	2100		0.01	1	1280	38	2	4		0.01	< 10	< 10	23	< 10	246	
197042	205 226	1650		0.01	3	840	68	6	2		0.01	< 10	< 10	13	< 10	348	
197043 197044	205 226	3830		0.01	5	1510	18	2	9		0.01	< 10	< 10	28	< 10	246	
197044 197045	205 226 205 226	1410 1510		0.01	3	1110	24	8	3		0.01	< 10	< 10	17	< 10	186	
197045	205 226	1510	4 (	0.01	2	1170	36	8	4	24 (	0.01	< 10	< 10	20	< 10	224	
197046	205 226	2550	1 <	0.01	3	1590	6	8	6	12 <	0.01	< 10	< 10	43	< 10	114	
197047	205 226	2810	1	0.01	1	1510	38	6	6	11 <	0.01	< 10	< 10	42	< 10	206	
197048	205 226	2510	< 1 <		1	155 <b>0</b>	2	6	7		0.01	< 10	< 10	56	< 10	86	
197049	205 226	1860	< 1	0.01	1	1310	2	12	6		0.01	< 10	< 10	39	< 10	120	
197050	205 226	2130	< 1 <	0.01	3	1370	14	16	4	28 <	0.01	< 10	< 10	22	< 10	122	
197051	205 226	1205		0.01	4	1350	14	12	6	22 <		< 10	< 10	30	< 10	128	
197052	205 226	6030	< 1 <		3	1340	2	6	9		0.01	< 10	< 10	56	< 10	46	
197053	205 226	6070		0.01	1	1390	6	6	8		0.01	< 10	< 10	58	< 10	94	
197054	205 226	4240	< 1 <		1	1600	12	12	6		0.01	< 10	< 10	40	< 10	96	
197055	205 226	6070	1 (	0.01	1	1450	6	8	8	31 <	0.01	< 10	< 10	49	< 10	64	
197056	205 226	2840	1 <	0.01	3	1450	16	14	6	28 <	0.01	< 10	< 10	27	< 10	156	
197057	205 226	3750		0.01	2	1230	16	8	7	19 <	0.01	< 10	< 10	38	< 10	70	
197058	205 226	6690		0.01	< 1	1300	10	6	7		0.01	< 10	< 10	43	< 10	104	
197059	205 226	2640		0.01	1	1300	18	12	5		0.01	< 10	< 10	27	< 10	176	
197060	205 226	2830	< 1 <	0.01	1	1650	2	2	. 7	22 <	0.01	< 10	< 10	55	< 10	84	
197061	205 226	1490	1 <	0.01	2	1550	50	14	2	32 <	0.01	< 10	< 10	22	< 10	266	<b>m</b>
197062	205 226	2180	< 1 <		< 1	1490	< 2	2	6	28 <	0.01	< 10	< 10	46	< 10	82	
197063	205 226	3680	< 1 <		1	1480	10	4	6		0.01	< 10	< 10	34	< 10	142	
197064	205 226	3220	< 1 <		1	1570	12	6	4		0.01	< 10	< 10	21	< 10	224	
197065	205 226	2850	2 <	0.01	1	1430	30	8	5	29 <	0.01	< 10	< 10	29	< 10	184	
197066	205 226	3370	< 1 <	0.01	1	1150	6	< 2	7	30 <	0.01	< 10	< 10	43	< 10	100	
197067	205 226	2280		0.01	4	1320	18	10	5		0.01	< 10	< 10	32	< 10	128	
197068	205 226	3220	< 1 <		3	1550	< 2	< 2	6		0.01	< 10	< 10	42	< 10	98	
197069	205 226	3830		0.01	3	1450	20	4	9		0.01	< 10	< 10	43	< 10	618	
197070	205 226	3690	1 <	0.01	3	1410	8	8	8	24 <	0.01	< 10	< 10	41	< 10	142	
197071	205 226	2820	3 <	0.01	4	1110	24	8	7	26 <	0.01	< 10	< 10	54	< 10	284	······································
197072	205 226	4000	6 <	0.01	6	1030	24	6	8	22 <	0.01	< 10	< 10	45	< 10	908	
197073	205 226	2750		0.01	4	1170	6	10	8		0.01	< 10	< 10	50	< 10	86	
197074	205 226	3270		0.01	9	850	< 2	10	9		0.01	< 10	< 10	44	< 10	102	
197075	205 226	1955	2 <	0.01	< 1	1530	4	24	5	15 <	0.01	< 10	< 10	21	< 10	80	
197076	205 226	1465	143 <	0.01	1	1390	14	6	5	14 <	0.01	< 10	< 10	28	< 10	140	
197077	205 226	1580	791 <		< 1	1090	30	12	3	9 <	0.01	< 10	< 10	15	< 10	124	
197078	205 226	1300		0.01	3	1210	12	14	4		0.01	< 10	< 10	22	< 10	138	
197079	205 226	2370		0.01	1	1630	2	6	7		0.01	< 10	< 10	59	< 10	72	
197080	205 226	380	< 1 <	0.01	< 1	20	6	< 2	< 1	10 <	0.01	< 10	< 10	< 1	< 10	8	
				. <u>.</u>		<u> </u>											



#### **Chemex Labs Ltd.**

Analytical Chemists \* Geochemists \* Registered Assayers

212 Brooksbank Ave., North Vancouver British Columbia, Canada V7J 2C1 PHONE: 604-984-0221 FAX: 604-984-0218 To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST. VANCOUVER, BC V6B 1N2

Page ....oer :3-A Total Pages :4 Certificate Date: 21-/ UG-97 0423

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Project : WKM 97-06

Comments: ATTN:S.HARRIS/J.LEHTINEN

**CERTIFICATE OF ANALYSIS** 

A9720423	
Invoice No. P.O. Number Account	: EIA

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SAMPLE	PREP CODE	Au ppb FA+AA	Au FA g/t	Ag ppm	Al %	As ppm	Ba ppm	Be ppm	Bi ppm	Ca %	Cd ppm	Co ppm	Cr ppm	Cu ppm	Fe %	Ga ppm	Hg Mgg	K S	La ppm	Mg %
197081	205 226	10 -		2.4	1.14	32	80	1.5	< 2	3.19	< 0.5	20	15	47	4.20	< 10	< 1	0.48	< 10	0.89
197082	205 226	< 5 -		0.2	1.32	70	10	1.5	< 2	0.76	< 0.5	29	11	18	3,62	< 10	< 1	0.30	10	0.25
197083	205 226	95 -		1.8	0.89	752	20	1.0	< 2	0.58	< 0.5	11	20	2	2.40	< 10	< 1	0.42	< 10	0.17
197084	205 226			1.4	0.65	142	< 10	0.5	< 2	0.09	< 0.5	< 1	37	1	0.31	< 10	< 1	0.33	< 10	0.03
197085	205 226	25 -	• •	3.8	0.55	46	< 10	0.5	< 2	0.22	< 0.5	< 1	34	10	0.39	< 10	< 1	0.30	< 10	0.07
197086	205 226	10 -		3.2	0.58	20	< 10	0.5	< 2	0.08	< 0.5	< 1	42	10	0.27	< 10	< 1	0.31	< 10	0.02
197087	205 226	5 -		0.8	0.64	36	< 10	0.5	< 2	0.08	< 0.5	< 1	41	5	0.27	< 10	< 1	0.34	< 10	0.02
197088	205 226	10 -		1.4	0.67	34	< 10	0.5	< 2	0.07	< 0.5	< 1	54	4	0.27	< 10	< 1	0.34	< 10	0.03
197089	205 226	10 -		0.8	0.54	30	< 10	0.5	< 2	0.06	< 0.5	< 1	39	1	0.25	< 10	< 1	0.30	< 10	0.01
197090	205 226	55 -		0.6	0.66	120	< 10	0.5	< 2	0.07	< 0.5	< 1	68	1	0.30	< 10	< 1	0.35	< 10	0.02
197091	205 226			< 0.2	0.55	28	< 10	0.5	< 2	0.09	< 0.5	< 1	56	< 1	0.28	< 10	< 1	0.31	< 10	0.03
197092	205 226	575 -		4.4	0.80	1295	< 10	0.5	< 2	0.40	< 0.5	6	39	8	1.03	< 10	< 1	0.48	< 10	0.13
197093	205 226	210 -		3.0	0.66	892	< 10	0.5	< 2	0.70	< 0.5	10	29	12	2.06	< 10	< 1	0.36	< 10	0.21
197094	205 226	1620	1.58	0.2	1.09	16	30	0.5	< 2	2.16	1.0	22	28	31	5,21	< 10	< 1	0.26	< 10	0.80
197095	205 226	300 -		0.6	1.40	62	40	0.5	< 2	1.19	1.5	22	13	47	5.85	< 10	< 1	0.47	< 10	0.57
197096	205 226	< 5 -		0.8	1.46	114	10	2.0	< 2	0.80	< 0.5	27	17	35	7.18	< 10	< 1	0.27	< 10	0.51
197097	205 226	< 5 -		0.2	1.07	114	30	1.0	< 2	1.85	< 0.5	20	14	62	3.89	< 10	< 1	0.53	10	0.62
197098	205 226	30 -		0.4	1.04	282	40	0.5	< 2	0.44	1.5	23	20	30	5.34	< 10	< 1	0.54	< 10	0.29
197099	205 226	30 -		0.8	0.92	240	50	0.5	< 2	0.31	2.0	21	20	37	4.70	< 10	< 1	0.47	< 10	0.08
197100	205 226	680 -		7.2	0.88	2350	40	0.5	< 2	0.35	2.0	23	25	45	5.87	< 10	< 1	0.43	< 10	0.06
197101	205 226	1680	1.61	42.4	0.69	1835	50	0.5	< 2	0.25	1.5	19	35	29	4.33	< 10	< 1	0.34	< 10	0.04
197102	205 226	1365	1.23	56.8	0.87	2690	40	0.5	< 2	0.26	1.5	18	40	24	5.08	< 10	< 1	0.39	< 10	0.04
197103	205 226	2340	2.33	11.0	0.69	5370	30	0.5	< 2	0.25	2.0	23	26	52	5.00	< 10	< 1	0.40	< 10	0.06
197104	205 226			3.0	1.13	232	50	1.5	< 2	0.51	0.5	23	21	78	5,26	< 10	< 1	0.57	< 10	0.29
197105	205 226	1700	1.68	50.0	0.63	2640	10	0.5	< 2	0.15	5.0	16	49	34	3.65	< 10	< 1	0.27	< 10	0.03
197106	205 226			3.0	1.03	246	30	1.0	< 2	0.30	2.0	24	20	76	4.94	< 10	< 1	0.44	< 10	0.10
197107	205 226	< 5 -		0.2	1.10	16	30	0.5	< 2	2.19	1.0	22	29	31	5.27	< 10	< 1	0.26	< 10	0.81
197108	205 226	10 -		0.6	1.43	66	40	0.5	< 2	1.23	1.5	22	12	48	5.96	< 10	< 1	0.47	< 10	0.58
197109	205 226	250 -		1.8	0.82	1385	30	1.0	< 2	2.70	1.5	17	14	22	5.23	< 10	< 1	0.40	< 10	0.81
197110	205 226	< 5 -		< 0.2	1.21	14	170	0.5	< 2	2.02	< 0.5	18	10	26	4.96	< 10	< 1	0.37	< 10	0.66
197111	205 226	15 -		< 0.2	0.87	32	50	0.5	< 2	0.42	1.5	19	15	13	5.49	< 10	< 1	0.50	< 10	0.39
197112	205 226	160 -		1.0	1.32	332	20	1.0	< 2	2.51	< 0.5	21	-9	35	5.28	< 10	$\langle \overline{1} \rangle$	0.39	< 10	0.56
197113	205 226	25 -		1.0	2,34	32	160	0.5	< 2	2.89	< 0.5	19	15	27	4.70	< 10	- À Î	0.39	< 10	1.20
197114	205 226	< 5 -		< 0.2	1.78	6	190	0.5	< 2	2.20	< 0.5	23	26	64	4.81	< 10	<1	0.51	10	0.88
197115	205 226	140 -		0.6	0.94	120	50	0.5	< 2	2.91	< 0.5	18	12	21	4.83	< 10	< 1	0.39	< 10	0.74
197116	205 226	< 5 -		< 0.2	1.11	12	70	0.5	< 2	0.75	< 0.5	29	17	59	5.38	< 10	< 1	0.47	< 10	0.38
197117	205 226	< 5 -		< 0.2	0.79	32	50	0.5	< 2	0.38	< 0.5	20	15	33	4.66	< 10	< 1	0.31	< 10	0.33
197118	205 226	< 5 -		< 0.2	0.97	30	50	0.5	< 2	0.38	1.5	23	19	33	5.44	< 10	< 1	0.42	10	0,29
197119	205 226	< 5 -		0.2	0.87	32	20	0.5	< 2	0.46	0.5	24	18	68	5.69	< 10	< 1	0.31	10	0,22
197120	205 226	< 5 -		< 0.2	1.01	10	10	1.0	< 2	0.51	< 0.5	18	17	60	4.75	< 10	< 1	0.29	10	0.31
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### **Chemex Labs Ltd.**

Analytical Chemists \* Geochemists \* Registered Assayers

212 Brooksbank Ave., North Vancouver British Columbia, Canada V7J 2C1 PHONE: 604-984-0221 FAX: 604-984-0218 fo: EQUITY ENGINEERING LTD.

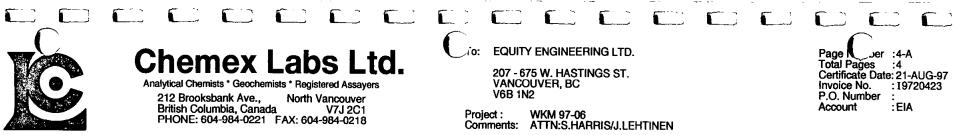
207 - 675 W. HASTINGS ST. VANCOUVER, BC V6B 1N2

Page Jer :3-B Total Pages :4 Certificate Date: 21-AUG-97 Invoice No. : 19720423 P.O. Number : Account : EIA

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Project : WKM 97-06 Comments: ATTN:S.HARRIS/J.LEHTINEN

									CERTIFICATE OF ANALYSIS			YSIS	Α	9720423				
SAMPLE	PREP CODE	Mn ppm	Mo ppm	Na %	Ni ppm	P Ppm	Pb ppm	Sb ppm	Sc ppm	Sr ppm	Ti %	Tl ppm	U PPm	V ppm	W mqq	Zn ppm		
197081	205 226	2210		0.01	4	1210	8	8	8	49 <	0.01	< 10	< 10	55	< 10	76		
197082	205 226	2710		0.01	3	1440	10	6	6		0.01	< 10	10	44	< 10	112		
197083	205 226	790		0.01	1	680	12	8	2		0.01	< 10	< 10	12	< 10	170		
197084 197085	205 226 205 226	65 185		(0.01 (0.01	< 1 < 1	20 10	8 10	< 2 4	< 1 < 1		0.01 0.01	< 10 < 10	< 10 < 10	< 1 < 1	< 10 < 10	12 18		
197086	205 226	35	< 1 <	0.01	< 1	10	10	4	< 1	24 <	0.01	< 10	< 10	< 1	< 10	14		
197087	205 226	35		0.01	< 1	10	12	2	< 1	25 <	0.01	< 10	< 10	< 1	< 10	14		
197088	205 226	30		0.01	< 1	< 10	8	< 2	< 1		0.01	< 10	< 10	< 1	< 10	12		
197089 197090	205 226	30 25		0.01 0.01	$\langle 1 \\ \langle 1 \rangle$	< 10 < 10	10 14	2			0.01	< 10	< 10	< 1	< 10	8		
	203 220		· · · ·	. 0.01	<u> </u>		14	< 2	< 1	24 <	0.01	< 10	< 10	< 1	< 10	18		
197091	205 226	80		0.01	< 1	< 10	2	< 2	< 1		0.01	< 10	< 10	< 1	< 10	2		
197092	205 226	545		0.01	3	220	10	6	2		0.01	< 10	20	12	< 10	74		
197093 197094	205 226 205 226	790 3410		0.01	5	160	12	8	5		0.01	< 10	10	15	< 10	106		
197095	205 226	3410		0.01	9 3	920 1210	62 56	2 6	13 7		0.01 0.01	< 10 < 10	< 10	97	< 10	312		
			<u> </u>			1210	50	0	/	10 (	0.01	10	< 10	63	< 10	368		
197096	205 226	9440		0.01	11	820	6	8	13	60 <	0.01	< 10	< 10	90	< 10	38	······	
197097	205 226	2850		0.01	8	910	2	16	7		0.01	< 10	< 10	35	< 10	58		
197098	205 226	3810		0.01	10	950	30	8	9		0.01	< 10	< 10	53	< 10	370		
197099 197100	205 226 205 226	1395 2890		0.01	7	980	40	12	5		0.01	< 10	< 10	34	< 10	376		
137100	203 220	2030	2 \	0.01	9	1120	34	24	4	8 <	0.01	< 10	< 10	26	< 10	258		
197101	205 226	1560	3 <	0.01	7	830	36	60	3	7 <	0.01	< 10	< 10	19	< 10	288		• • • • • • • • • • • • • • • • • • • •
197102	205 226	635	4 <	0.01	8	910	56	64	3		0.01	< 10	< 10	20	< 10	260		
197103	205 226	1735		0.01	9	810	54	22	5	7 <	0.01	< 10	< 10	26	< 10	428		
197104	205 226	5060		0.01	9	1240	26	20	12		0.01	< 10	< 10	60	< 10	244		
197105	205 226	310	4 <	0.01	8	440	202	34	2	6 <	0.01	< 10	< 10	25	< 10	656		
197106	205 226	1920	8 <	0.01	10	610	84	22	6	18 <	0.01	< 10	< 10	45	< 10	460		
197107	205 226	3450		0.01	10	910	64	2	13	20 <	0.01	< 10	< 10	97	< 10	314		
197108	205 226	3800		0.01	1	1200	54	4	7		0.01	< 10	< 10	64	< 10	380		
197109 197110	205 226	4330		0.01	< 1	1320	28	8	8		0.01	< 10	< 10	32	< 10	266		
19/110	205 226	2200	< I (	0.01	< 1	1210	2	2	8	45 (	0.01	< 10	< 10	66	< 10	116		
197111	205 226	4690		0.01	7	940	20	2	12	8 <	0.01	< 10	< 10	60	< 10	344		
197112	205 226	2050		0.01	3	1480	14	6	7		0.01	< 10	< 10	52	< 10	124		
197113	205 226	1860	< 1		1	1430	10	2	6	78	0.03	< 10	< 10	81	< 10	96		
197114 197115	205 226 205 226	1685 3190		0.01	10	1320	6	< 2	7		0.01	< 10	< 10	78	< 10	104		
	203 220	2120	<u>`</u> I (	0.01	< 1	1470	12	6	6	43 <	0.01	< 10	< 10	32	< 10	78		
197116	205 226	2010	2 <	0.01	12	860	2	< 2	4	12 <	0.01	< 10	< 10	47	< 10	92		
197117	205 226	2360		0.01	9	1010	6	4	6		0.01	< 10	< 10	44	< 10	78		
197118	205 226	2870		0.01	8	900	6	6	8		0.01	< 10	< 10	51	< 10	314		
197119 197120	205 226 205 226	2780 2560		0.01	8	1210	6	14	8		0.01	< 10	< 10	45	< 10	98		
12/120	203 220	₹30V	<u>, </u> , , , , , , , , , , , , , , , , , ,	0.01	6	1190	8	14	11	11 (	0.01	< 10	< 10	69	< 10	66		
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#### **Chemex Labs Ltd.**

Analytical Chemists \* Geochemists \* Registered Assayers 212 Brooksbank Ave., North Vancouver British Columbia, Canada V7J 2C1 PHONE: 604-984-0221 FAX: 604-984-0218

Cio: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST. VANCOUVER, BC V6B 1N2

Page Jer :4-A Total Pages :4 Certificate Date: 21-AUG-97 Invoice No. : 19720423 P.O. Number ٠ Account :EIA

Project : WKM 97-06 Comments: ATTN:S.HARRIS/J.LEHTINEN Project :

									······		CE	ERTIFI	CATE	OF A	NAL	YSIS		9720	423		
SAMPLE	PRE COD		Au ppb FA+AA	Au FA g/t	Ag ppm	Al %	As ppm	Ba ppm	Be ppm	Bi ppm	Ca %	Cd ppm	Co ppm	Cr ppm	Cu ppm	Fe %	Ga ppm	Hg ppm	K S	La ppm	Mg %
197121 197122 197123 197124 197125	205 205 205 205 205 205	226 226 226	10 < 5 30		0.4 0.2 < 0.2 < 0.2 < 0.2 0.6	0.83 0.80 0.72 0.53 0.75	188 20 22 12 94	50 30 30 10 40	0.5 0.5 0.5 < 0.5 0.5	<pre> &lt; 2 &lt; 2 &lt; 2 &lt; 2 &lt; 2 &lt; 2 &lt; 2 &lt; 2 &lt; 2 &lt; 2</pre>	0.26 0.19 0.40	< 0.5 < 0.5 < 0.5 < 0.5 < 0.5 < 0.5	11 19 18 13 31	24 10 12 19 11	35 54 22 9 76	2.84 5.89 7.06 3.66 5.80	<pre>&lt; 10 &lt; 10 &lt; 10 &lt; 10 &lt; 10 &lt; 10 &lt; 10 &lt; 10</pre>	<pre>&lt; 1 &lt; 1 &lt; 1 &lt; 1 &lt; 1 &lt; 1 &lt; 1 &lt; 1 &lt; 1 &lt; 1</pre>	0.32 0.18 0.20 0.18 0.18	10 10 < 10 < 10 < 10	0.04 0.08 0.13 0.21 0.50
197126 197127 197128 197129	205 205 205 205 205	226 226	165 25		< 0.2 0.8 13.4 < 0.2	1.29 0.82 7.72 2.91	8 1890 32 16	40 50 190 90	< 0.5 0.5 < 0.5 < 0.5 < 0.5	<pre>&lt; 2 &lt; 2 &lt; 2 &lt; 2 &lt; 2 &lt; 2 &lt; 2</pre>	2.71 2.78	< 0.5 < 0.5 16.0 < 0.5	23 25 25 16	22 18 28 28	26 17 4060 50	5.92 5.42 6.97 4.81	< 10 < 10 10 < 10	<pre>&lt; 1 &lt; 1 &lt; 1 &lt; 1 &lt; 1 &lt; 1 &lt; 1</pre>	0.19 0.29 1.86 0.28	< 10 < 10 < 10 < 10 < 10	0.91 0.82 2.05 1.00



### Chemex Labs Ltd.

Analytical Chemists \* Geochemists \* Registered Assayers 212 Brooksbank Ave., North Vancouver

British Columbia, Canada V7J 2C1 PHONE: 604-984-0221 FAX: 604-984-0218

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST. VANCOUVER, BC V6B 1N2

Project : WKM 97-06 Comments: ATTN:S.HARRIS/J.LEHTINEN

CEDTIFICATE OF ANALVEIS A0700400

Total Pages : Certificate Date: Invoice No. : P.O. Number :	4-B 4 21-AUG-97 19720423 EIA
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SAMPLE	PR CO		Mn ppm	Мо ррт	Na %	Ni ppm	р ррт	РЬ ррт	Sb ppm	Sc ppm	Sr ppm	Ti %	Tl ppm	U ppm	V ppm	W Ppm	Zn ppn	
197121 197122 197123 197124 197125	205 205 205	226 226 226 226 226 226	585 920 1040	1 1 1 ( 1	0.01 0.08 0.04 0.04 0.03	3 5 8 3 10	500 670 220 170 790	12 8 6 12	14 16 4 < 2 16	1 2 4 5 7	27 < 18 < 15 <	0.01 0.01 0.01 0.01 0.01 0.01	<pre>&lt; 10 &lt; 10 &lt; 10 &lt; 10 &lt; 10 &lt; 10 &lt; 10</pre>	<pre>&lt; 10 &lt; 10 &lt; 10 &lt; 10 &lt; 10 &lt; 10 &lt; 10 &lt; 10</pre>	9 37 27 31 40	< 10 < 10 < 10 < 10 < 10 < 10	98 60 38 30 84	
197126 197127 197128 197129	205 205	226 226 226 226	1870 2250	1 1 2 < 1	< 0.01 0.42	5 4 < 1 < 1	870 1280 1060 2150	8 20 490 8	< 2 6 4 < 2	4 6 7 5		0.01 0.01 0.15 0.10	< 10 < 10 < 10 < 10	< 10 < 10 < 10 < 10	30 40 152 91	< 10 < 10 < 10 < 10	68 90 1610 90	

#### **APPENDIX E**

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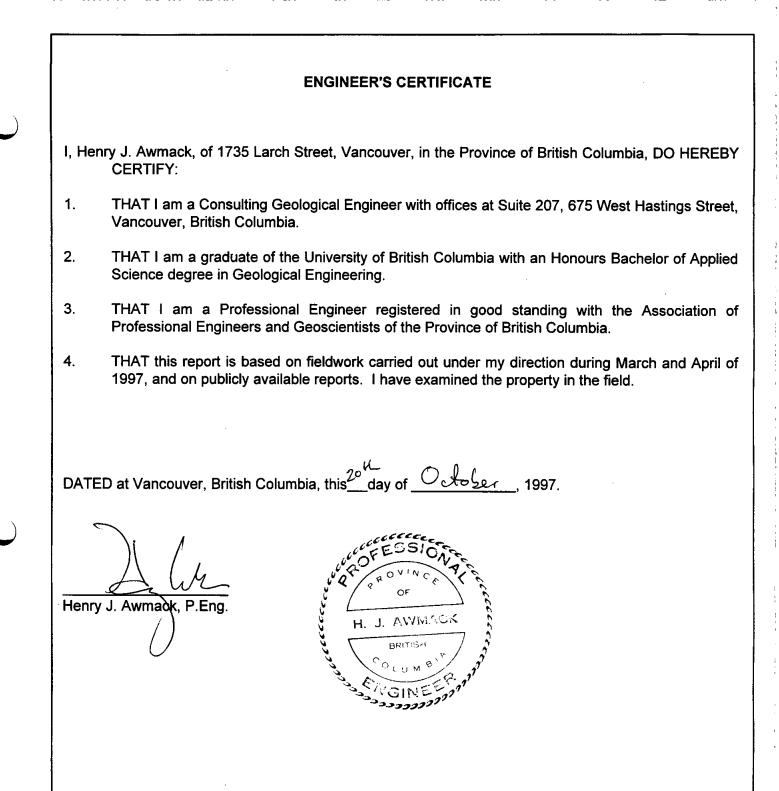
#### **GEOLOGIST'S AND ENGINEER'S CERTIFICATES**

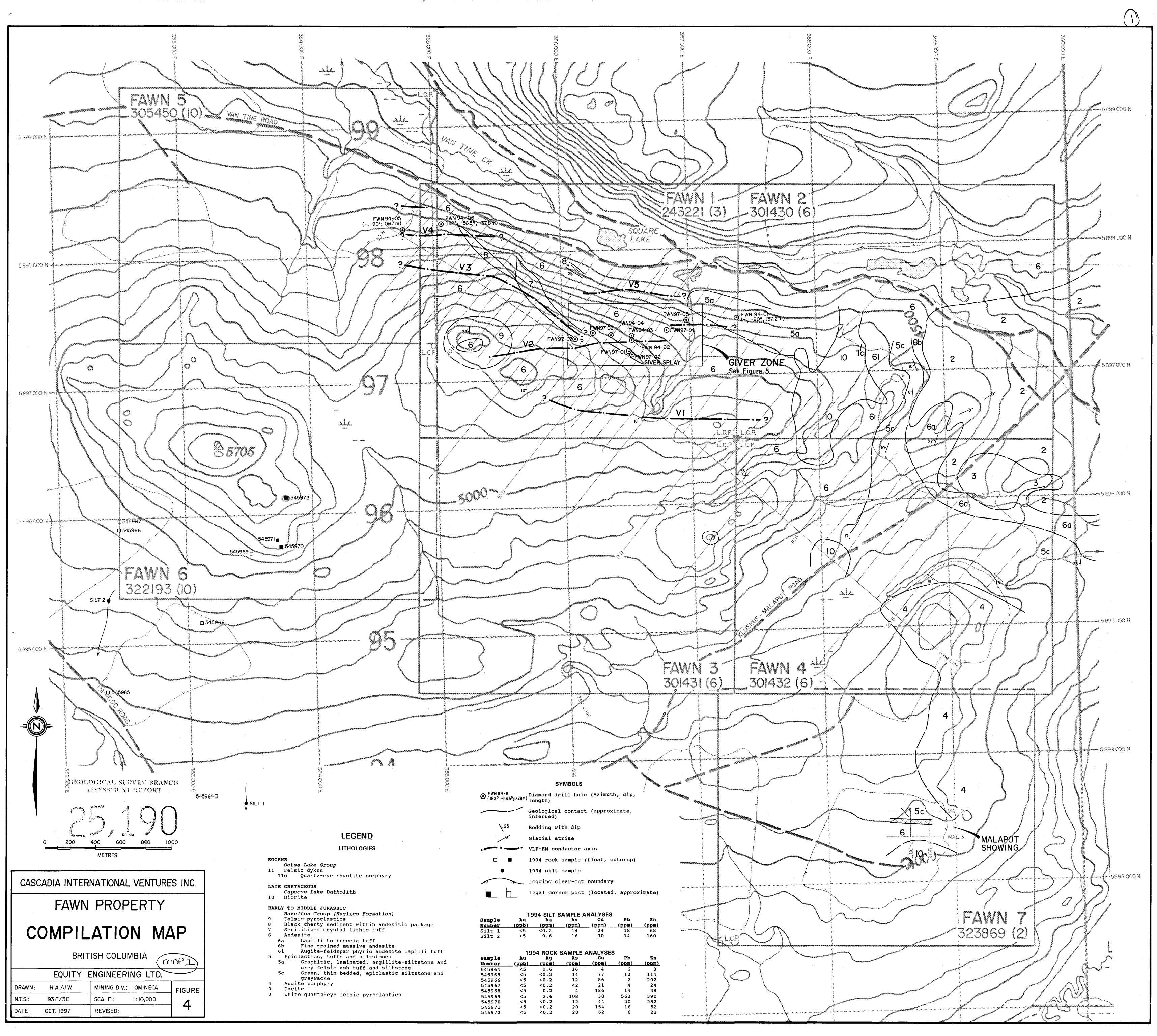
#### **GEOLOGIST'S CERTIFICATE**

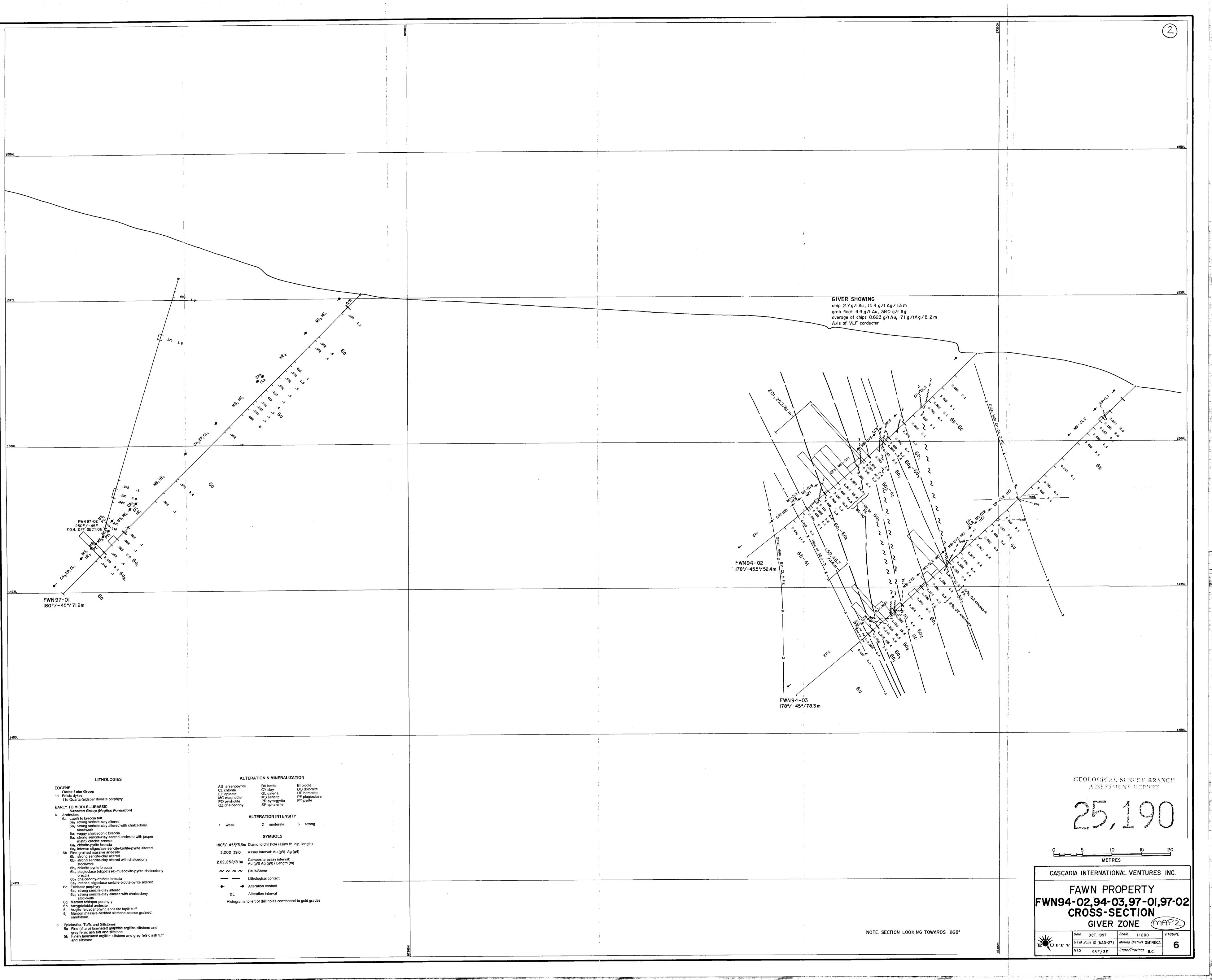
- I, Jim Lehtinen, of 4317 Briardale Road, Royston in the Province of British Columbia, DO HEREBY CERTIFY:
- 1. THAT I am a Consulting Geologist with Equity Engineering Ltd. with offices at Suite 207, 675 West Hastings Street, Vancouver, British Columbia.
- 2. THAT I am a graduate of the University of British Columbia with a Bachelor of Science degree in Geology.
- 3. THAT I am a Professional Geoscientist registered in good standing with the Association of Professional Engineers and Geoscientists of the Province of British Columbia.
- 4. THAT this report is based on a diamond drilling program I supervised in March and April of 1997, and on publicly available reports.

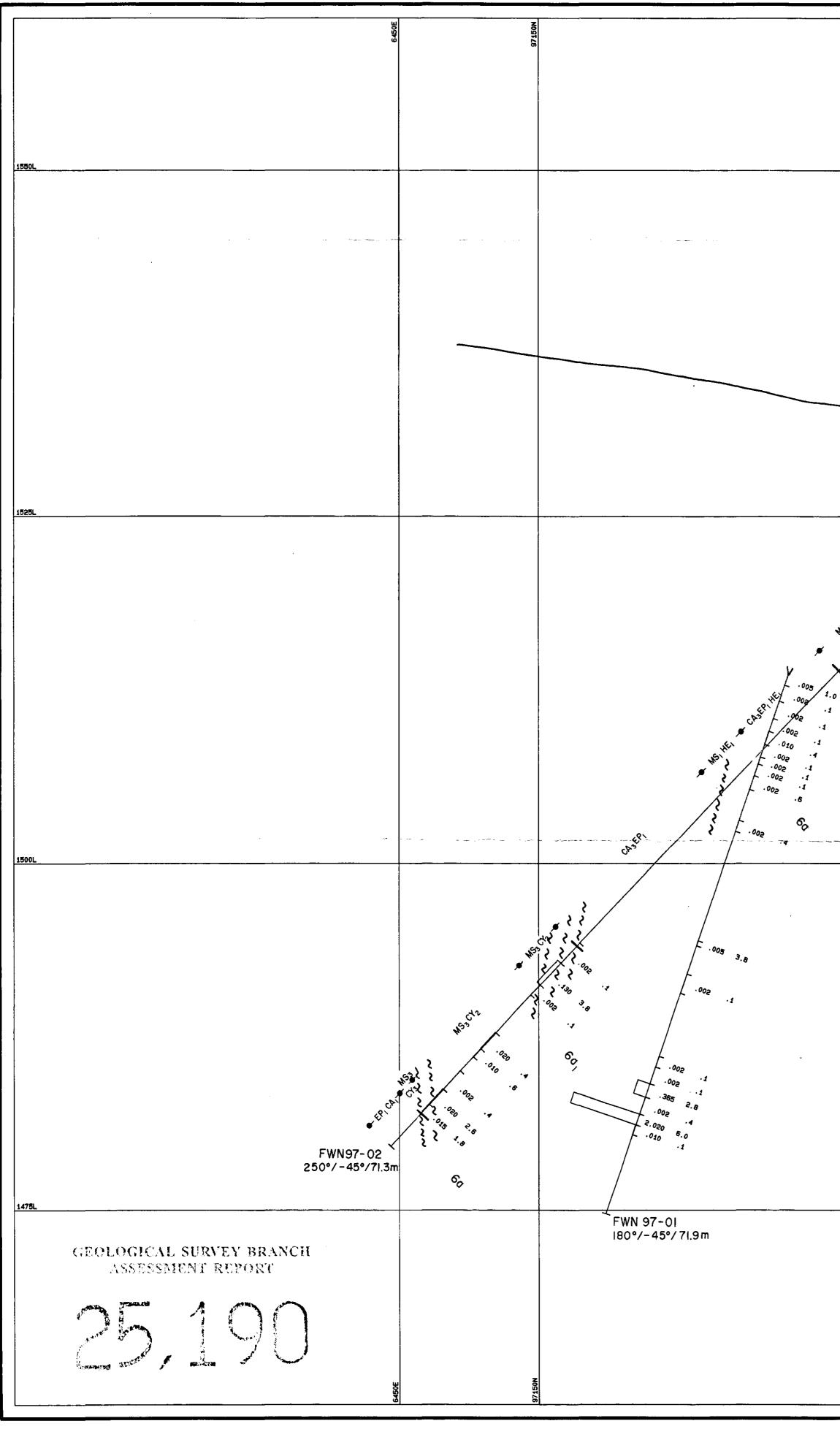
DATED at Vancouver, British Columbia, this \_\_\_\_ day of October, 1997.

Jim Lehtinen, P.Geo.









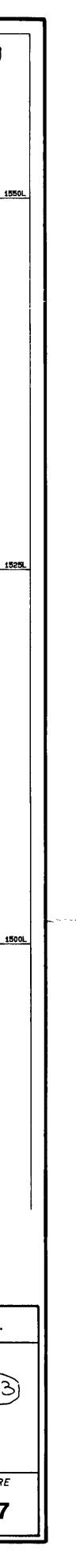
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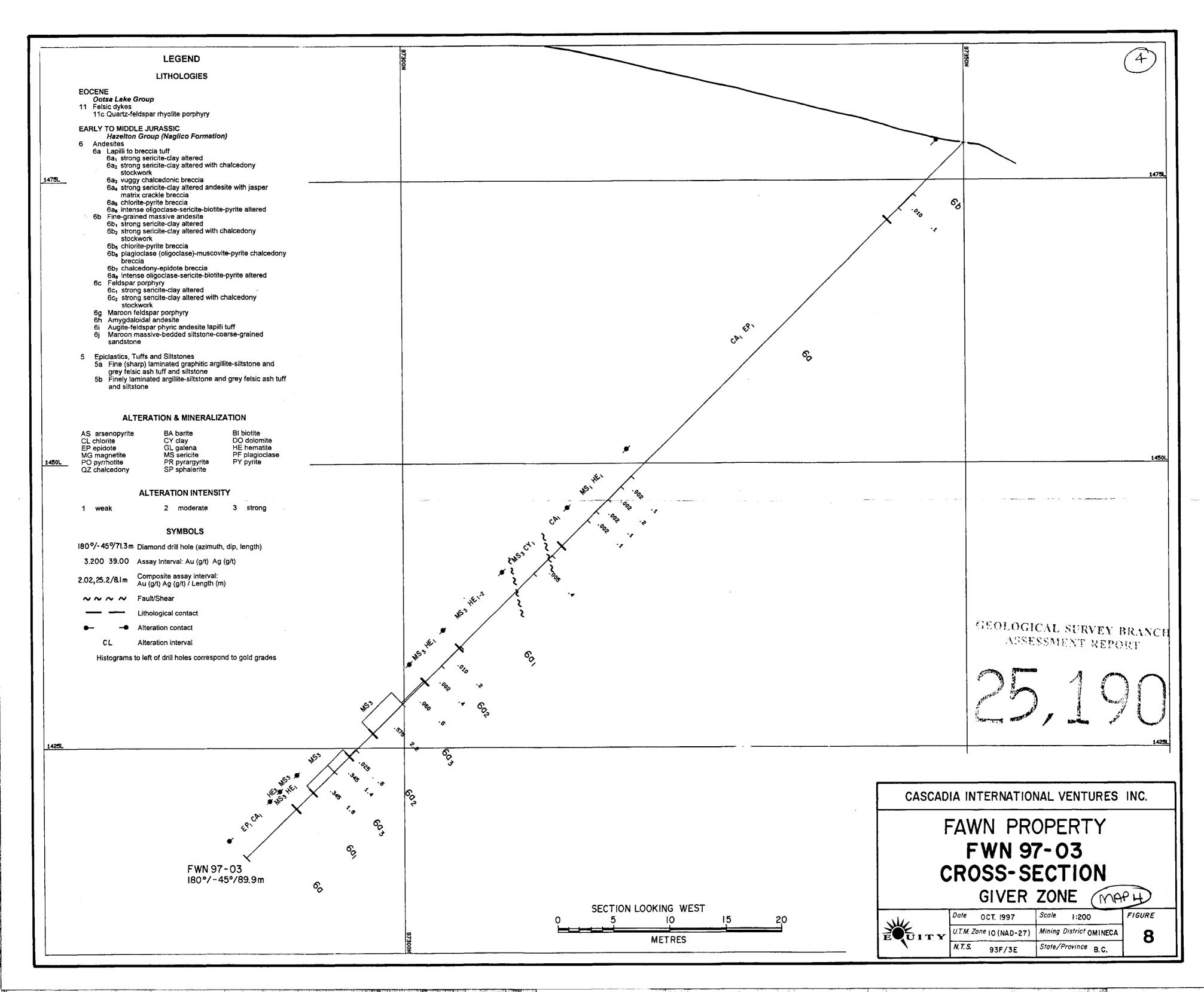
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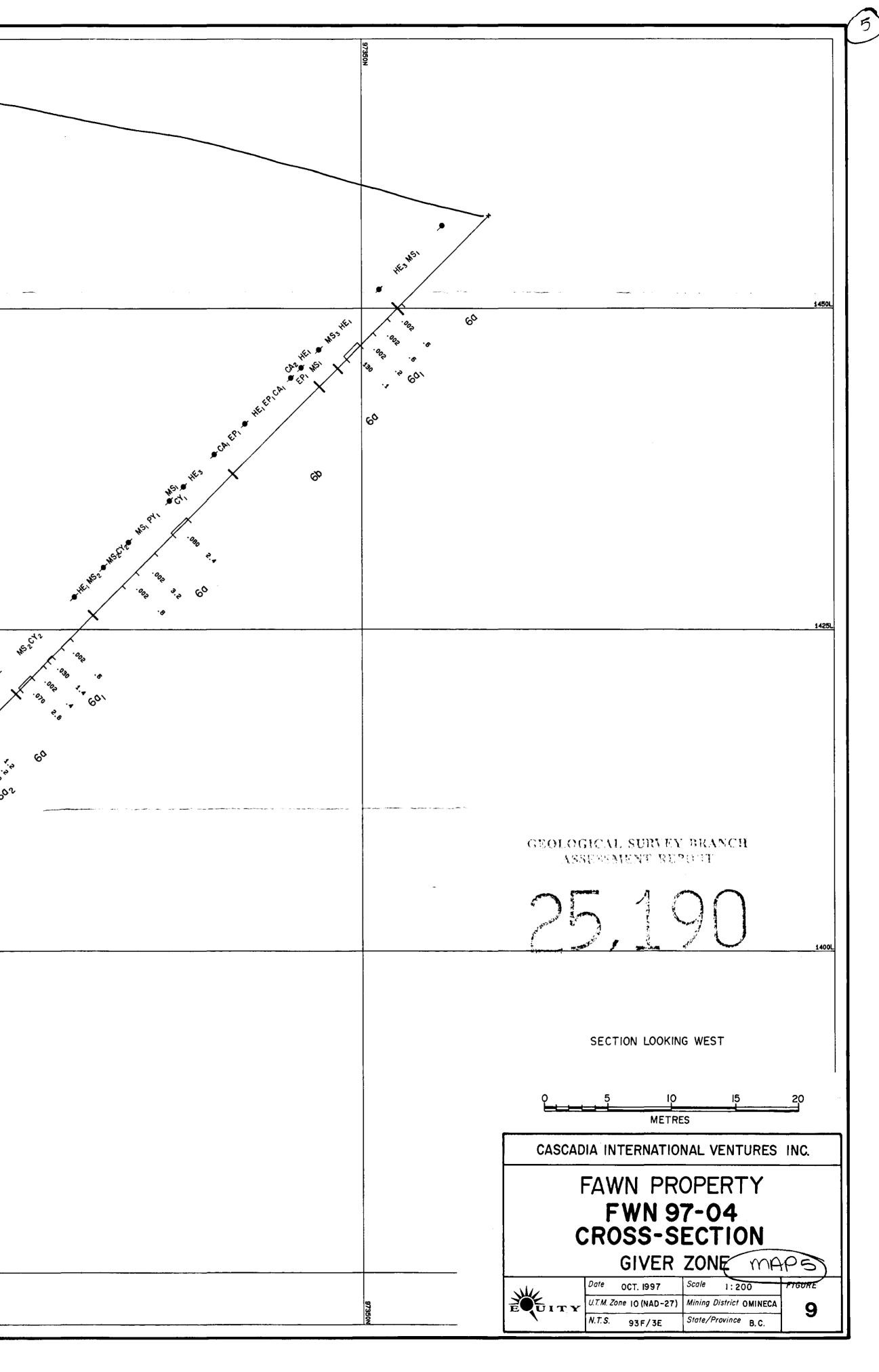
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			LEGEND	
	<b>、</b>		EOCENE <i>Ootsa Lake Group</i> 11 Felsic dykes 11c Quartz-feldspar rhyolite porphyry	
			EARLY TO MIDDLE JURASSIC Hazelton Group (Naglico Formation) 6 Andesites 6a Lapilli to breccia tuff 6a1 strong sericite-clay altered	
			<ul> <li>6a2 strong sericite-clay altered with chalcedony stockwork</li> <li>6a3 vuggy chalcedonic breccia</li> <li>6a4 strong sericite-clay altered andesite with jasper matrix crackle breccia</li> </ul>	
NS2 <sup>HE</sup>			6a <sub>5</sub> chlorite-pyrite breccia 6a <sub>8</sub> intense oligoclase-sericite-biotite-pyrite altered 6b Fine-grained massive andesite 6b <sub>1</sub> strong sericite-clay altered 6b <sub>2</sub> strong sericite-clay altered with chalcedony	
CA3EP1		,,,,, <del>,,,,,,,,,,,,,,,,,,,,,,,,,,,,,</del>	<ul> <li>Stockwork</li> <li>6b<sub>5</sub> chlorite-pyrite breccia</li> <li>6b<sub>6</sub> plagioclase (oligoclase)-muscovite-pyrite chalce breccia</li> <li>6b<sub>7</sub> chalcedony-epidote breccia</li> </ul>	dony
× ~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~			6a₅ intense oligoclase-sericite-biotite-pyrite altered 6c Feldspar porphyry 6c₁ strong sericite-clay altered 6c₂ strong sericite-clay altered with chalcedony	
NE314:2 H. 13			stockwork 6g Maroon feldspar porphyry 6h Amygdaloidal andesite 6i Augite-feldspar phyric andesite lapilli tuff 6j Maroon massive-bedded siltstone-coarse-grained sandstone	
€g			<ul> <li>5 Epiclastics, Tuffs and Siltstones</li> <li>5a Fine (sharp) laminated graphitic argillite-siltstone and grey felsic ash tuff and siltstone</li> <li>5b Finely laminated argillite-siltstone and grey felsic ash and siltstone</li> </ul>	i 1 tuff
			ALTERATION & MINERALIZATION	
			AS arsenopyriteBA bariteBI biotiteCL chloriteCY clayDO dolomiteEP epidoteGL galenaHE hematiteMG magnetiteMS sericitePF plagiocla:PO pyrrhotitePR pyrargyritePY pyriteQZ chalcedonySP sphalerite	
			ALTERATION INTENSITY 1 weak 2 moderate 3 strong	
			I80%-45%71.3m Diamond drill hole (azimuth, dip, length)	
			3.2003.9.0Assay Interval: Au (g/t) Ag (g/t)2.02,25.2/8.1mComposite assay interval: Au (g/t) Ag (g/t) / Length (m)	
			Fault/Shear     Eithological contact	
			<ul> <li>Alteration contact</li> <li>CL Alteration interval</li> <li>Histograms to left of drill holes correspond to gold grades</li> </ul>	s
			I SECTION LOOKING 340°	
			0 5 IO IS METRES	20
			CASCADIA INTERNATIONAL VENTU	IRES INC.
			FAWN PROPERT FWN 97-02	(MAP 3
			CROSS-SECTIO GIVER ZONE	N
6500E			Date OCT. 1997 Scale 1: 200 U.T.M. Zone 10 (NAD-27) Mining District Of	
8  8	l	L	N.T.S. 93F/3E State/Province	B. C.

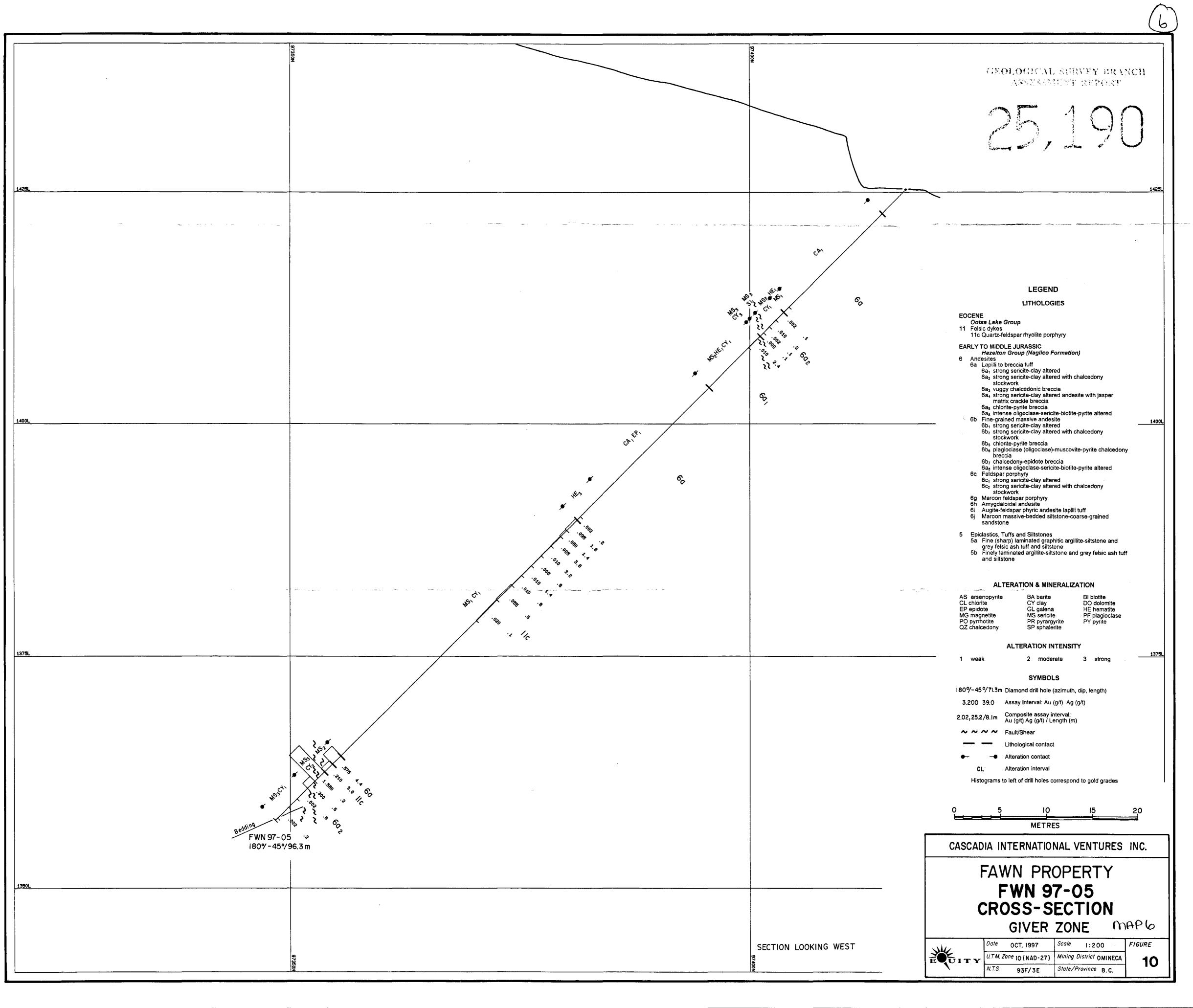
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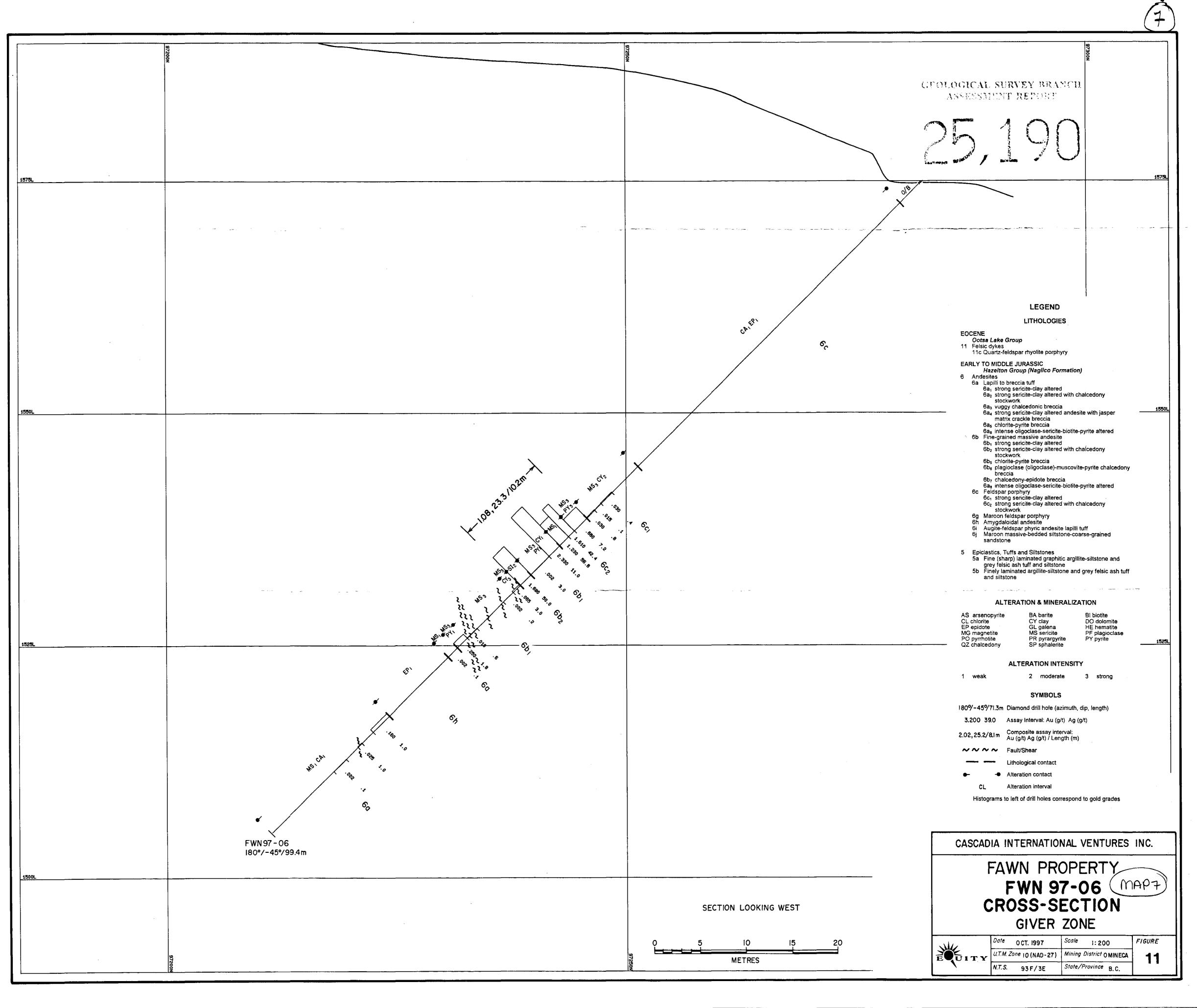




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	LITHOLOGIES	
	EOCENE Ootsa Lake Group	
	11 Felsic dykes 11c Quartz-feldspar rhyolite porphyry	
	EARLY TO MIDDLE JURASSIC Hazelton Group (Naglico Formation)	
	6 Andesites 6a Lapilli to breccia tuff	
	6a₁ strong sericite-clay altered 6a₂ strong sericite-clay altered with chalcedony	
	stockwork 6a3 vuggy chalcedonic breccia	
	6a4 strong sericite-clay altered andesite with jasper matrix crackle breccia	
	6a₅ chlorite-pyrite breccia 6a₅ intense oligoclase-sericite-biotite-pyrite altered 6b Fine-grained massive andesite	
1450L	6b <sub>1</sub> strong sericite-clay altered 6b <sub>2</sub> strong sericite-clay altered 6b <sub>2</sub> strong sericite-clay altered with chalcedony	
	stockwork 6b5 chlorite-pyrite breccia	
	6bs plagioclase (oligoclase)-muscovite-pyrite chalcedony breccia	
	6b7 chalcedony-epidote breccia 6a8 intense oligoclase-sericite-biotite-pyrite altered	
	6c Feldspar porphyry 6c1 strong sericite-clay altered	
	6c₂ strong sericite-clay altered with chalcedony stockwork 6g_ Maroon feldspar porphyry	
	6h Amygdaloidal andesite 6i Augite-feldspar phyric andesite lapilli tuff	
	6j Maroon massive-bedded siltstone-coarse-grained sandstone	
	5 Epiclastics Tuffs and Siltstones	
	5a Fine (sharp) laminated graphitic argillite-siltstone and grey felsic ash tuff and siltstone 5b Finely laminated argillite-siltstone and grey felsic ash tuff	
	5b Finely laminated argillite-siltstone and grey felsic ash tuff and siltstone	
	ALTERATION & MINERALIZATION	
	AS arsenopyrite BA barite BI biotite	
	CL chlorite CY clay DO dolomite EP epidote GL galena HE hematite	
	MG magnetite MS sericite PF plagioclase PO pyrhotite PR pyrargyrite PY pyrite QZ chalcedony SP sphalerite	
	QZ chalcedony SF sphalente	
	ALTERATION INTENSITY	
1425L	1 weak 2 moderate 3 strong	
	SYMBOLS	
	180°/-45°/7I.3m Diamond drill hole (azimuth, dip, length)	
	3.200 39.0 Assay Interval: Au (g/t) Ag (g/t)	
	2.02, 25.2/8.1m Composite assay interval: Au (g/t) Ag (g/t) / Length (m)	EP HENST
	$\sim \sim \sim \sim$ Fault/Shear	MON NO.
	Lithological contact	N52
	Alteration contact	NS2 33
	CL Alteration interval	
	Histograms to left of drill holes correspond to gold grades	N <sup>52</sup> 7 <sup>2</sup> 2 <sup>3</sup> 2 <sup>3</sup> 2 <sup>3</sup> 2 <sup>3</sup> 2 <sup>3</sup> 2 <sup>3</sup> 2 <sup>3</sup> 2
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	LEGEND			
	LITHOLOGIES			
	EOCENE Ootsa Lake Group			
	11 Felsic dykes 11c Quartz-feldspar rhyolite porphyry			
	EARLY TO MIDDLE JURASSIC Hazelton Group (Naglico Formation)			
	6 Andesites 6a Lapilli to breccia tuff			
1600L	6a, strong sericite-clay altered 6a <sub>2</sub> strong sericite-clay altered with chalcedony stockwork			
	6a₃ vuggy chalcedonic breccia 6a₄ strong sericite-clay altered andesite with jasper matrix crackle breccia			
	6as chlorite-pyrite breccia 6as intense oligoclase-sericite-biotite-pyrite altered			
	6b Fine-grained massive andesite 6b1 strong sericite-clay altered 6b2 strong sericite-clay altered with chalcedony	, , <sup>1</sup> and the transformed and the second		
	stockwork 6bs chlorite-pyrite breccia 6be plagioclase (oligoclase)-muscovite-pyrite chalcedony			
	breccia 6by chalcedony-epidote breccia			
	6a, intense oligoclase-sericite-biotite-pyrite altered 6c Feldspar porphyry 6c1 strong sericite-clay altered			
	6c <sub>2</sub> strong sericite-clay altered with chalcedony stockwork			
	6g Maroon feldspar porphyry 6h Amygdaloidal andesite 6i Augite-feldspar phyric andesite lapilli tuff 6j Maroon massive-bedded siltstone-coarse-grained			
	6j Maroon massive-bedded siltstone-coarse-grained sandstone			
	5 Epiclastics, Tuffs and Siltstones 5a Fine (sharp) laminated graphitic argillite-siltstone and graph falsic ash fulf and siltstone			
	<ul> <li>5 Epicialitis, rule and onderlies</li> <li>5a Fine (sharp) laminated graphitic argillite-siltstone and grey felsic ash tuff and siltstone</li> <li>5b Finely laminated argillite-siltstone and grey felsic ash tuff and siltstone</li> </ul>			
	ALTERATION & MINERALIZATION			
1575L	AS arsenopyrite BA barite BI biotite CL chlorite CY clay DO dolomite EP epidote GL galena HE hematite			
	MG magnetite MS sericite PF plagioclase PO pyrrhotite PR pyrargyrite PY pyrite			
	QZ chalcedony SP sphalerite			
	ALTERATION INTENSITY			
	1 weak 2 moderate 3 strong			-
	SYMBOLS			\$ \$ \$
	1809-45971.3 m Diamond drill hole (azimuth, dip, length) 3.200 39.0 Assay Interval: Au (g/t) Ag (g/t)			S <sup>W</sup> 2,
	2.02, 25.2/8.Im Composite assay interval: Au (g/t) Ag (g/t) / Length (m)			* 3
	Au (g/t) Ag (g/t) / Lengur (m) → → → Fault/Shear	,		×.
	Lithological contact			153 CA 241 CL'
	Alteration contact			2 CA2 2
	CL Alteration interval Histograms to left of drill holes correspond to gold grades	د. میسیمی المیکند در این میانید برای میکند. ا		M <sup>3</sup>
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