


**ASSESSMENT REPORT
ON THE
ANDY 3 and 4 CLAIMS
Geochemical Survey**

Merritt Area, British Columbia
Nicola Mining Division
NTS: 092 I 27 & 37

Latitude: 50° 16' 30" N
Longitude: 120° 46' 00" W

FOR
Owners/Operator: Ahura Mining Ltd.

 **GEOLOGICAL SURVEY BRANCH**
ASSESSMENT REPORT

By Marthe Archambault, P. Geo
May, 2002

26,872

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1 INTRODUCTION

This report was written in order to fulfill the assessment work requirements.

Fieldwork, consisting of soil geochemistry, prospecting with some rock sampling was conducted from March 24th to 30th 2001.

2 LOCATION, ACCESS AND TOPOGRAPHY

The ANDY 3 and 4 claims are located approximately 25 km north of the town of Merritt and 5 km west of Swakum Mountain. They are centered at 50° 16' 30"N and 120° 46' 00"W, NTS map sheet 92I / 27 and 37 (Figures 1 and 2).

The property is accessible by taking the Helmer Lake Road exit on the Coquihalla Highway, approximately 26 km north of the town of Merritt. From Helmer Lake, a 8.2 km non-maintained gravel logging road leads to Eve Lake and was recently extended to the south through the ANDY 3 claim, reaching its northern boundary at km 11.3(from Helmer Lake).

The property is located within the Thompson Plateau, a physiographic division of the Interior Plateau System. Topography is typically gently rolling terrain with elevation relief averaging 200 metres from elevations of 1350 metres to 1550 metres. To the east of the property, Swakum Mountain dominates with a peak elevation of 1858 metres.

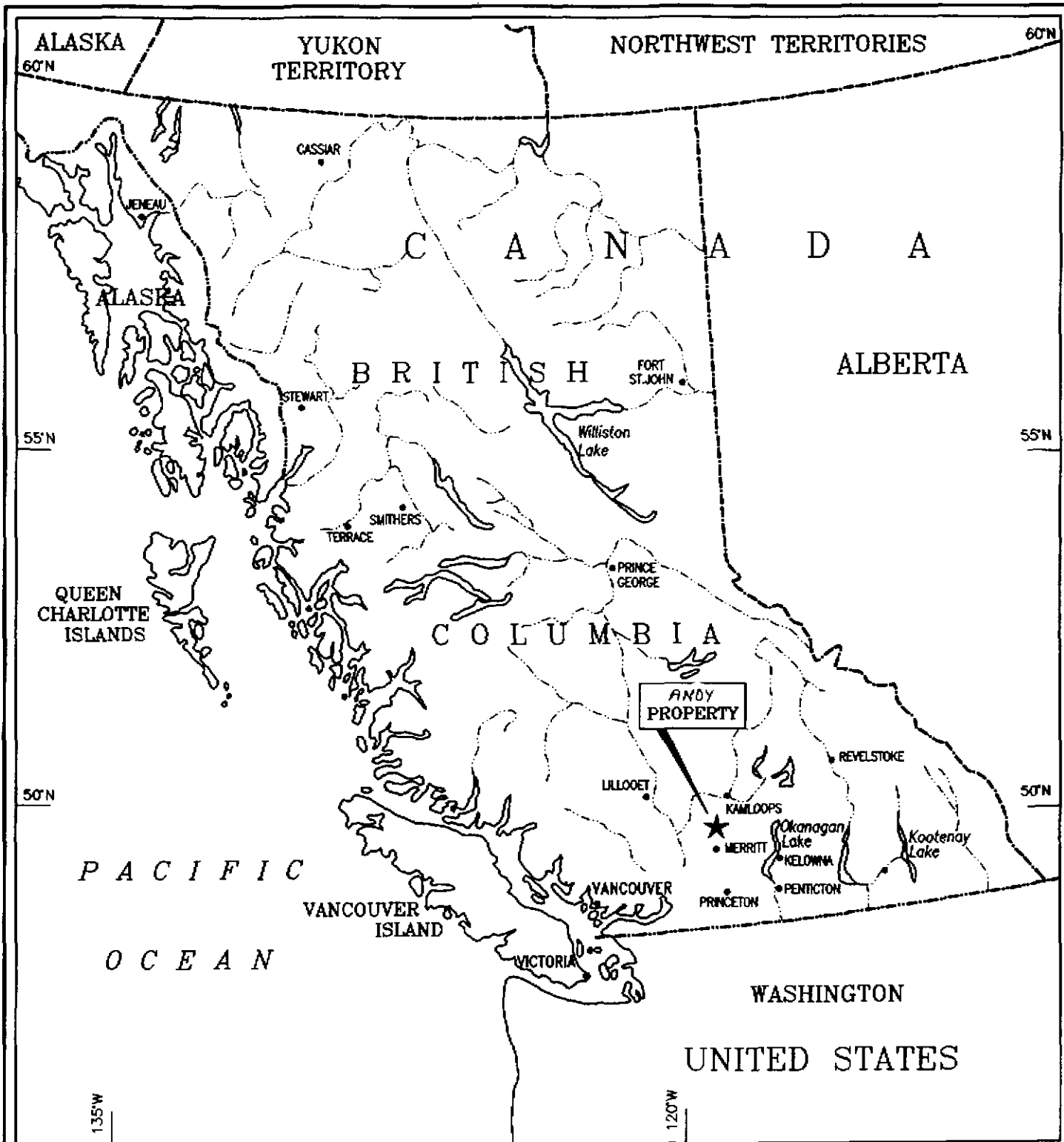
3 PROPERTY OWNERSHIP AND MINERAL TENURE

The property is part of the Nicola Mining Division, and owned by Ahura Mining Ltd. The two claims are composed of a total of thirty-six units.

A more detailed description of the mineral tenures is presented in Table 1.

TABLE 1 PROPERTY OWNERSHIP AND CLAIM TENURE

Tenure Number	Claim Name	Owner		Map Number	Work Recorded to	Units	Tag Number
		Number	%				
385428	ANDY 3	140388	100	092I037	March 24, 2004	16	210571
385429	ANDY 4	140388	100	092I027	March 27, 2004	20	210573
						36	



LEGEND

- CITY OR TOWN
- PROVINCIAL BOUNDARY
- INTERNATIONAL BOUNDARY
- RIVER



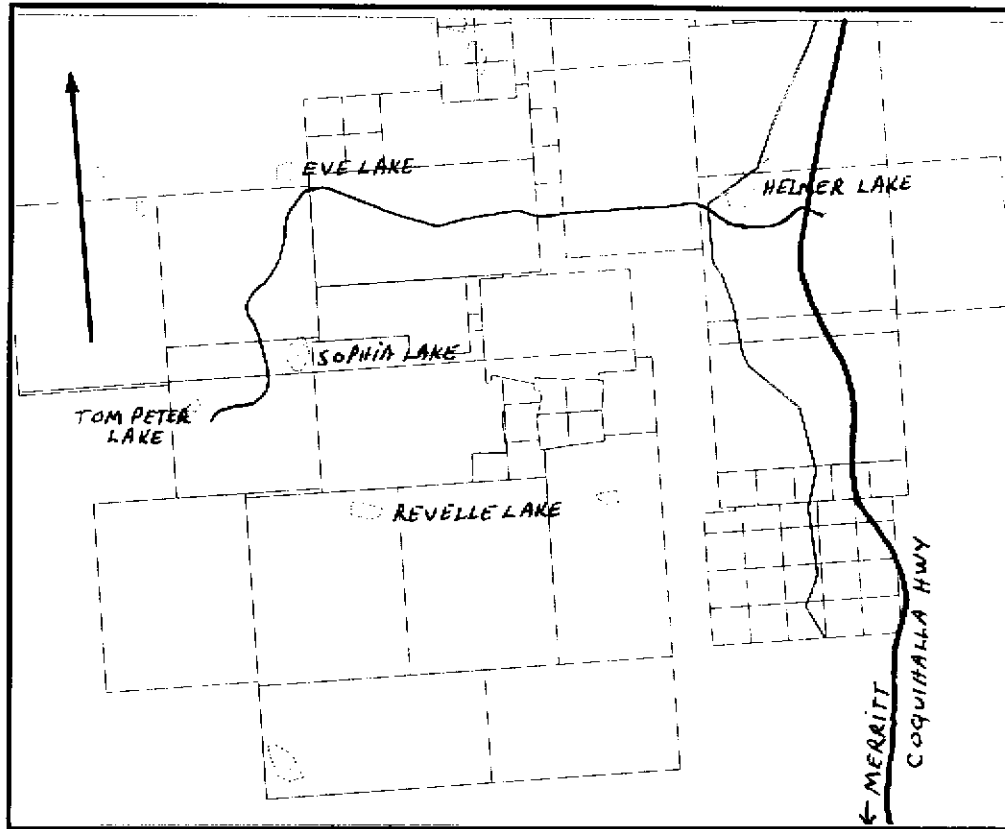
AHURA MINING LTD.

ANDY PROJECT
Nicola Mining Division, B.C.

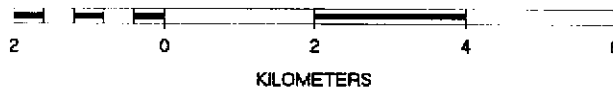
GENERAL LOCATION MAP



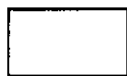



Project No:	By: M.A.
Scale: 1:8,000,000	Drawn: Alpha-2000 Drafting Mj
Date:	Figure: 1



SCALE 1 : 100,000

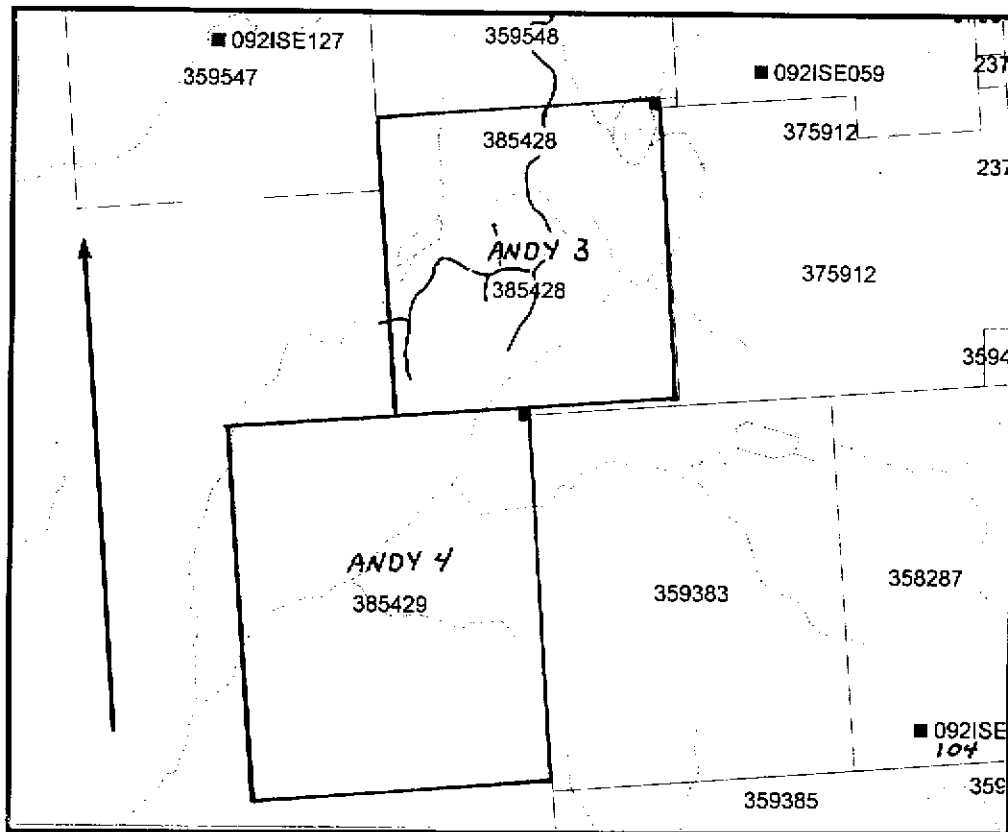


LEGEND

-  Claim Outline
-  Lake
-  Highway
-  Logging Road

after: http://ebony.gov.bc.ca/mapplace/maps/minpol/dep_find.MWF

AHURA MINING LTD.			
ANDY PROJECT			
Nicola Mining Division, B.C.			
PROPERTY ACCESS MAP			
Project No:		By:	
Scale:	1:100,000	Drawn by:	M.A.
Date:	May-02	Figure:	2



SCALE 1 : 50,000



LEGEND



Claim Outline
 Claim Name
 Tenure No.
 Legal Corner Post



Minfile Developed Prospect



Minfile Showing

092ISE 127

after: http://ebony.gov.bc.ca/mapplace/maps/minpot/dep_find.MWF

AHURA MINING LTD.			
ANDY PROJECT			
Nicola Mining Division, B.C.			
CLAIM LOCATION MAP			
Project No:		By:	
Scale:	1:50,000	Drawn by:	M.A.
Date:	May-02	Figure:	3

4 EXPLORATION HISTORY

The discovery of mineralization in the Swakum Mountain area dates back to 1916. Work on the Tolman Lake area, one kilometre to the northwest of ANDY 3 started in 1965.

The history of exploration in the ANDY claims area is summarized in Table 2.

TABLE 2 HISTORY OF EXPLORATION WORK

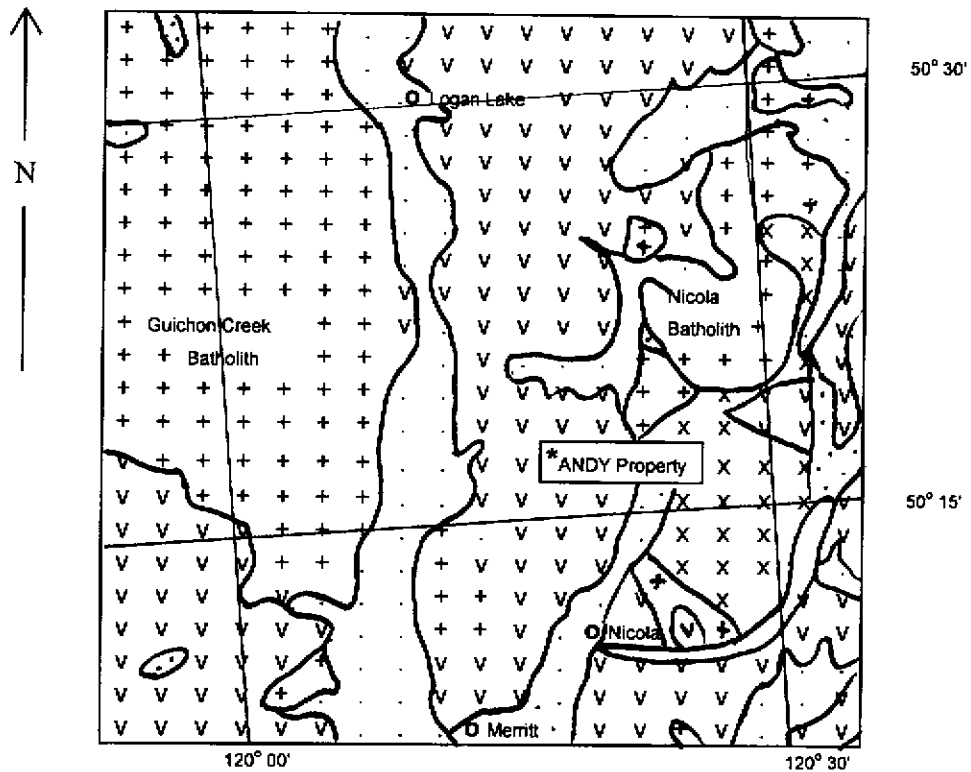
YEAR	COMPANY	LOCATION	TYPE OF WORK	PRODUCTION
1965-66	Vastlode Mining Co.	Tolman Lake area	Stripping, trenching, 17 holes of diamond drilling and limited magnetometer survey	
1967-68	San Doh Mines Ltd.	Tolman Lake area	Trenching, 34 holes of diamond drilling	
1969	Highland Lode Mines	Tolman Lake area	Geological mapping	
1971	Highland Lode Mines	Tolman Lake area	Resource calculation for Zone No. 3 by Mr. Elwell, P. Eng.	The arithmetic average grade of the zone was 0.36 oz Ag/ton; 1.69% Pb; 4.80% Zn; and 0.18% Cu.
1971-72	Highland Lode Mines	Tolman Lake area	Geological mapping, 6 holes of diamond drilling and road building	
1976	Ruskin Developments Ltd	Tolman Lake area	VLF-EM survey, geochemical survey	
1977	Highland Lode Mines	Tolman Lake area	stripping, trenching, 900 metres of diamond drilling in 16 holes, adit driving for 325 m.	
1977	Ruskin Developments Ltd.	Tolman Lake, Zone 3	Diamond Drilling, 210.3m in 4 holes	
1977-78 1979	Mr. C. Boitard and Lakewood Mining Co. Ltd.	Sophia Lake	14 km of Magnetometer, VLF surveys 11.8 km of IP and soil sampling for Pb, Zn, Cu, Ag, WO ₃ 3 shallow Diamond Drill Holes 548.6 metres of percussion drilling in 6 holes	
1979-1980	C.D.R. Resources Inc.	Tolman Lake area	587metres of diamond drilling in 12 holes	
1981 to 1985	Mr. Sherwin F. Kelly	West of Lucky Mike, west and south-west of Swakum Mtn	Soil geochemical surveys on several small grids in order to fulfill assessment work requirements. The samples were generally analyzed for Cu, Pb, Zn, Ag	
1983	Lakewood Mining Co. Ltd.	Sophia Lake	170 m. of percussion drilling in two holes	

YEAR	COMPANY	LOCATION	TYPE OF WORK	PRODUCTION
YEAR	COMPANY	LOCATION	TYPE OF WORK	PRODUCTION
1984-85-86	Mr. J. Georgilas, Decade International development Ltd.	South of Corona showing West of Corona	Geological mapping and magnetometer survey Two lines of soil geochemical survey with 82 samples analysed for Cu, Pb, Zn, Ag &As VLF-EM survey	
1986	Atlas Resources Ltd	Corona-Thelma	29.1 km, magnetometer, VLF-EM, geological mapping, soil geo-chemical survey with 424 samples analyzed for 29 elements plus silver and gold.	
1986-1987	Lakewood Mining Co. Ltd.	Sophia	4.2 km of magnetometer and VLF-EM survey and 5.1 km of induced polarization	
2001 to present	Ahura Mining Ltd.		Prospecting and geochemical surveys.	

5 REGIONAL GEOLOGY

The property is located within the Nicola Belt, a terrain approximately 40 kilometres wide and 180 kilometres long, extending from the International Boundary to the south to Kamloops Lake to the north. It consists mainly of Upper Triassic volcanic, sedimentary, and intrusive rocks of the Nicola Group.

In the property area, the Nicola Group forms a band approximately fifteen kilometres wide and sixty kilometres long (Figure 4). It is bound on its east side by the granodioritic rocks of the Early Jurassic Nicola Batholith and the dioritic rocks of a smaller intrusion. On its west side, it is flanked by the large Triassic Jurassic Guichon Creek Batholith. The Guichon and Nicola Batholiths were emplaced during the late stages of volcanic activity. They show evidence of having been intruded at shallow depth. Both are phased bodies and grade in composition from granodiorite to quartz-monzonite. The adjacent Nicola Group hosts numerous dykes and small plugs of similar composition.



LEGEND

- * Project Location
- o Community
- 1:50k Grid

LITHOLOGIES

- Alluvial
- ++ Guichon Creek Batholith
- x+ Multi-phased Nicola Batholith
- vv Nicola Volcanics

AHURA MINING LTD.			
ANDY PROJECT			
Nicola Mining Division, B.C.			
REGIONAL GEOLOGY			
Project No:		By:	
Scale:	1:500,000	Drawn by:	M.A.
Date:	May-02	Figure:	4

after: http://ebony.gov.bc.ca/mapplace/maps/minpol/dep_find.MWF

6 PROPERTY GEOLOGY

The Nicola Group rocks in the Swakum Mountain area generally strike north to northeast with generally steep dips. For a large part, they consist of andesitic flows and tuffs, agglomerates, and occasional basalts and rhyolites. In a few places, a break occurs in the volcanic stratigraphy and is comprised of a mixed volcanic-sedimentary unit consisting of a thick sequence of felsic volcanic flows, lithic and crystal tuffs, limy sediments and a prominent limestone. These units generally have a northeast strike and can extend for 2.5 kilometre in strike length. Narrow quartz porphyry dykes locally intrude the Nicola Group sequence. Most of the old workings on Swakum Mountain occur in close proximity to or at the volcanic-sedimentary contact. The Swakum Mountain deposits have been historically classified as polymetallic "skarn-type" mineralization, lead-zinc-silver bearing quartz veins and replacements, and polymetallic quartz veins. (after MINFILE #092ISE027)

There are a few historical mineral occurrences in the property surroundings. A short description of these occurrences, extracted from the government MINFILES, is given below, and their location is shown on Figure 3.

CORONA NO. 1 (L.4512) – Showing

MINFILE # 092I SE 104

COMMODITIES: Silver, Gold, Lead, Zinc

The Old Corona No. 1 occurrence is located within folded, fine to medium-grained interbedded basalt flows, tuffs and agglomerates of the Nicola Group. Vein-type mineralization occurs in a 25 to 70 metre wide zone of limonitic, ankeritic volcanics occurs in a distinct depression between prominent north-northwest trending ridges of unaltered volcanics.

In the collapsed Corona shaft and in several trenches, quartz-carbonate veins are 5 to 10 centimetres wide with a northwest strike and steep westward dip. Mineralization includes galena, sphalerite, pyrite and tetrahedrite. Samples from the Corona shaft assayed up to 2442.1 grams per tonne silver and 0.1 grams per tonne gold (Assessment Report 15312).

SOPHIA – Showing

MINFILE # 092I SE 059

COMMODITIES: Copper, Lead, Zinc

At the Sophia showing, mineralization occurs in a shear zone exposed in a trench. The zone is 8 metres wide, strikes 220 degrees and dips 30 to 60 degrees south. Pyrite, sphalerite, galena and chalcocopyrite are associated with quartz and calcite which occur as narrow stringers in andesitic porphyry and as cement in brecciated volcanics.

TOLMAN LAKE AREA

The Tolman Lake area is underlain by intermediate volcanoclastic and flow rocks of the Nicola Group. A strongly brecciated shear zone strikes 045 degrees and dips steeply to the northwest and is apparently continuous over a strike length of 2000 metres. This structure hosts 3 known zones of mineralization.

SUNSHINE ZONE 1- Showing

MINFILE # 092I SE 127

COMMODITIES: Lead, Zinc

Zone 1 is located on the west side of Tolman Lake. Open cuts expose a shear striking 085 degrees and dipping 65 degrees to the north within andesitic tuffs. The shear varies in width from 3 to 8 metres and contains steeply dipping quartz stringers which coalesce downward into a 30 to 60 centimetre wide vein. The vein is fractured and in places brecciated. The fractures are mineralized with sphalerite and minor galena.

SUNSHINE ZONE 2 - Prospect

MINFILE # 092I SE 128

COMMODITIES: Silver, Copper, Lead, Zinc

For 46 metres along the strike of the zone, four trenches expose brecciated andesitic tuffs mineralized with quartz, sphalerite, pyrite, chalcopyrite and galena. The quartz stringers and sulphides lie in two principal directions; one strikes east and dips 75 degrees north and the second strikes 045 degrees and dips 80 degrees southeast. The mineralized brecciated andesitic tuffs are cut by several unmineralized steep faults trending east and northwest.

Combined average assay results from diamond drilling were 0.157 per cent lead, 4.10 per cent zinc, 2.4 grams per tonne silver and 0.17 per cent copper over 3.25 metres; gold averaged less than 0.1 grams per tonne (Assessment Report 8036). Inferred resources are 45,359 to 54,431 tonnes based on a width of 8 metres, length of 50 metres and depth of 30 metres (Assessment Report 6742).

SUNSHINE ZONE 3 – Developed Prospect

MINFILE # 092I SE 129

COMMODITIES: Silver, Copper, Lead, Zinc

In Zone 3, galena, sphalerite, chalcopyrite, pyrite and pyrrhotite occur in a brecciated zone with a quartz-calcite matrix. The hangingwall consists of bleached and pyritic andesite which grades into numerous quartz and calcite veins carrying sphalerite and galena. The footwall consists of highly silicified andesite containing unmineralized quartz and calcite veins. The mineralized brecciated zone varies in width up to 6.5 metres and is cut and slightly offset by several near-vertical, north trending faults.

Zone 3 has been tested by diamond drilling and underground development to a depth of 50 metres over a length of 165 metres. In 1971, unclassified resources are 258,523 tonnes averaging 1.69 per cent lead, 4.8 per cent zinc, 0.18 per cent copper and 12.34 grams per tonne silver. The grade is difficult to determine due to very poor drill core recovery. The grade is based on drill core and adit sampling (Elwell, 1971).

A short geochemical sampling program was conducted by a prospector / technician, Ted Hayes, between March 24th and 30th, 2001.

Eighty-six (86) soil samples, of which seventy-six (76) are located within the claim boundaries, were collected from the B-horizon, at 50 metres intervals along a newly build logging road on ANDY 3 (Figure 5). The sample spacing was measured with a belt-chain and the sample locations marked with flagging tape. The position of the road and samples was entered on a Garmin 12XL handheld GPS (Global Positioning System). The sample numbers correspond to the established kilometre markers along the logging road. Thirteen (13) rock samples, of which twelve (12) are located within the claim boundaries, were taken from some of the outcrops along the road, with the sample number also corresponding to the distance along the logging road. All samples consist of altered Nicola volcanics with minor pyrite mineralization.

All samples were sent to Acme Analytical Laboratories Ltd. for 30 element ICP and gold by AA analysis. Both soil and rock samples were analysed as follows:

GROUP 1D: 0.50gm sample leached with 3mn 2-2-2 HCL-HNO3-H2O at 95 deg. C for one hour, diluted to 10 ml, analysed by ICP-ES (Atomic Emission Spectrometer). Upper limits – Ag, Au, Hg, W = 100 ppm; Mo, Co, Cd, Sb, Bi, Th, U and B = 2,000 ppm; Cu, Pb, Zn, Ni, Mn, As, V, La, Cr = 10,000 ppm. Gold by acid leached, analysed by ICP-MS (Mass Spectrometer) (10 gm).

Copper, lead, and zinc results are plotted on figure 6 and listed in Table 3 with gold results. Statistical analysis was not performed on the results given that the sample population is quite small. For the time being, an arbitrary threshold of 70ppm for copper, 25 ppm for lead and 100ppm for zinc are considered above background. Worth mentioning are rock sample #13775 and the adjacent soil sample # 13770. They both indicate strong copper-lead-zinc-gold mineralization with high silver and cadmium.

TABLE 3 SAMPLE RESULTS

ROCK SAMPLES

ELEMENT	Cu	Pb	Zn	Au*
SAMPLES	ppm	ppm	ppm	ppb
12000R	10	3	33	1.9
12830	55	< 3	49	2.4
12840	48	< 3	59	3.4
13001	41	< 3	39	0.4
13528	7	10	61	1.3
13530	21	< 3	157	2.0
13590	58	< 3	107	2.4
13775	1751	26	3015	2.4
1300301	46	27	197	3.6
1305001	47	3	56	2.2
1305002	45	8	50	4.1
1320400	48	< 3	83	1.5
1400200	134	< 3	94	1.5

SOIL SAMPLES

ELEMENT	Cu	Pb	Zn	Au*
SAMPLES	ppm	ppm	ppm	ppb
11000	42	7	78	3.0
11050	36	11	84	2.4
11100	60	< 3	94	2.0
11150	35	7	89	1.4
11200	52	7	70	1.8
11250	71	5	83	3.1
11300	57	11	86	3.6
11350	42	5	70	2.4
11400	40	8	63	3.1
11450	60	8	77	4.2
11500	48	9	84	3.3
11550	61	8	71	2.7
11600	42	3	92	4.2
11650	56	3	59	3.4
11700	32	4	67	0.8
11750	75	7	94	3.3
11800	50	7	77	5.1
11850	44	9	72	1.1
11900	59	7	86	2.4
11950	46	7	65	1.1
12000	29	8	72	0.7
12000A	63	6	61	1.6
12600	33	5	69	2.2
12650	46	10	70	2.3
12700	58	8	92	2.5
12750	35	8	60	2.0
12800	67	11	67	5.7
12850	74	22	114	4.2
12900	37	8	87	1.2
12950	49	10	80	4.1
1275050	42	< 3	58	1.6
1275100	49	< 3	50	2.7
1275150	51	3	62	2.1
1275200	66	6	68	2.9
1275250	61	14	77	3.1

ELEMENT SAMPLES	Cu ppm	Pb ppm	Zn ppm	Au* ppb
1275300	34	10	78	1.3
1275350	53	8	74	1.9
1275400	30	7	73	0.9
1275450	36	8	62	6.3
1275500	30	5	59	2.1
14000	34	13	70	4.0
14050	38	15	89	2.7
14100	56	20	101	5.6
14150	43	29	129	2.5
14200	41	38	110	1.3
14250	58	26	131	3.0
14300	50	27	116	2.2
14350	71	20	112	4.6
14400	72	24	111	3.7
14450	57	25	161	1.2
1400050	24	21	127	0.5
1400100	23	13	95	1.0
1400150	21	8	66	0.6
1400200	38	16	79	1.6
1400250	38	11	76	2.4
1400300	18	9	99	4.2
1320050	41	13	78	1.1
1320100	40	15	80	6.5
1320150	30	10	81	2.1
1320200	58	16	71	3.6
13000	31	8	75	0.9
1300050	52	9	67	3.6
1300100	56	8	57	3.0
1300150	38	6	53	3.2
1300200	30	8	68	1.5
1300250	28	8	74	1.2
1300300	32	9	75	1.3
1300350	34	11	70	1.9
13150	34	10	70	1.4
13200	38	15	88	1.2
13250	37	12	85	1.6
13300	50	14	94	4.0
13350	67	12	88	3.9
13400	65	42	154	6.9
13450	63	16	125	2.0
13500	35	15	103	1.4
13550	49	29	129	3.0
13600	76	35	143	10.4
13650	59	9	73	10.9
13700	74	16	103	3.9
13750	51	10	123	3.4
13800	56	13	121	3.1
13850	53	18	109	3.2
13900	38	27	117	3.1
13950	64	39	115	4.4
13770	2400	359	4773	671.7

8 CONCLUSIONS

The ANDY property lies within a favourable geological environment where the volcanic package of the Triassic Nicola Group has been compressed on both sides by two granodioritic batholiths causing regional folding and fracturing of the Nicola sequence. This prepared the ground for hydrothermal activity and mineral deposition. The Nicola volcanics are also a favourable environment for a Volcanic Massive Sulphide Deposit.

Exploration/Mining activity in the Swakum Mountain area dates back to the 1900's. Work to date has shown that several of the historical workings contain small deposits of high-grade copper, lead, zinc with some silver, gold and tungsten. The ANDY property hasn't been systematically explored. The 2001 geochemical survey has returned significant results which should be investigated. The ANDY 4 claims remains unexplored.

9 RECOMMENDED WORK PROGRAM

The 2001 geochemical sampling was successful in finding a new zone of mineralization. However, it only covered a small portion of ANDY 3. It is therefore recommended to follow up with a small \$50,000 exploration program with the objective of defining the new mineralized zone and to continue the grass-roots exploration over the remaining of the ANDY 3 claim and all of the ANDY 4. Some line cutting and soil sampling on lines 500m apart would give an initial coverage and can be in-filled to a 100 metre spacing during the next phase depending upon the results. The soil samples will be taken from the B-horizon and analyzed by the 30 element ICP method and gold geochemistry Atomic Absorption method. Reconnaissance geological mapping, prospecting and rock sampling should be carried out at the same time.

Further work will be contingent on the results of this initial work phase.

PROPOSED BUDGET FOR THE PHASE I EXPLORATION PROGRAM

The cost of the proposed work program is summarized below in Canadian Funds for a period of fifteen days.

Mobilization/Demobilization	\$	3,500
Camp and Support	\$	5,600
Truck Rental	\$	2,000
Grid Preparation and Line Cutting	\$	4,400
Soil Geochemical Survey	\$	13,718
Geological Mapping	\$	9,200
Rock Sampling	\$	800
Expendables	\$	1,000
Communication	\$	235
Technical Report	\$	5,000
Contingency	\$	<u>4,545</u>
TOTAL	\$	50,000

10 COST STATEMENT

The 2001 fieldwork program was carried out between March 24th and March 30th, 2001.
The cost is detailed as follows:


Room & Board:	3 days @ \$125/day	\$ 375
Off-Road 4x4 Truck Rental, incl. Unlimited km and insurance,	4 days @ \$150/day	\$ 600
Technician/Prospector, Ted Hayes	5 days @ \$350/ day	\$ 1,750
Soil Geochemical Sampling	76 samples @ \$18/sample	\$ 1,368
Rock Geochemical Sampling	12 samples @ \$25/sample	\$ 300
Communications		\$ 25
Miscellaneous & Consumables		\$ 190
Report		<u>\$ 1,500</u>
TOTAL OF FIELD EXPENDATURES		\$ 6,108
Withdrawal from the PAC account		\$ 1,092
TOTAL VALUE OF WORK		\$ 7,200

11 CERTIFICATE OF QUALIFICATIONS

I, Marthe Archambault, of Surrey, British Columbia, hereby certify that:

- I am a Consulting Geologist with an office at #1601 – 13880 101st Avenue, Surrey, British Columbia.
- I graduated with a Bachelor's degree in Geology from the University of Montreal, Montreal, Quebec (1980) and a Master's degree in Geology from the University of British Columbia, Vancouver, British Columbia (1985).
- I am a member of the Association of Professional Engineers and Geoscientists of British Columbia, Registration No. 19226.
- I have practiced my profession as an exploration geologist for the past 22 years.
- I have visited the ANDY claims and inspected the sampling location.
- I have no interest, direct or indirect, nor do I expect to receive any interest, in the ANDY property or Ahura Mining Ltd.

Dated at Surrey, British Columbia this 19th day of June 2002.



Marthe Archambault, P. Geo.

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B.C. Ministry of Energy and Mines MINFILE Database: 092ISE 027, 059, 094, 095, 096, 097, 098, 099, 100, 101, 102, 103, 104, 105, 106, 127, 128, 129, 160.

The Map Place Website: http://ebony.gov.bc.ca/mapplace/minpot/min_titl.cfm

APPENDIX I
ANALYTICAL RESULT
LABORATORY CERTIFICATE



GEOCHEMICAL ANALYSIS CERTIFICATE

Hayes, T. File # A100861 Page 1
1704 - 555 Austin Ave, Coquitlam BC V3K 6R8 Submitted by: Ted Hayes

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm	Au* ppb
11000	<1	42	7	78	.3	21	15	723	3.94	4	<8	<2	2	41	<.2	<3	<3	119	.90	.053	8	39	1.02	195	.15	4	2.74	.02	.09	<2	3.0
11050	<1	36	11	84	<.3	16	13	866	3.54	<2	<8	<2	2	38	.2	<3	3	93	1.11	.056	8	28	.77	185	.14	3	2.88	.02	.10	<2	2.4
11100	<1	60	<3	94	.3	8	20	1563	5.51	6	<8	<2	<2	34	.2	<3	<3	121	2.19	.100	6	13	1.26	94	.27	5	3.24	.02	.08	<2	2.0
11150	1	35	7	89	<.3	17	14	518	3.85	3	<8	<2	2	35	<.2	<3	3	115	.57	.055	7	33	.85	183	.15	4	3.07	.01	.11	<2	1.4
11200	1	52	7	70	<.3	21	17	491	4.32	7	<8	<2	<2	40	<.2	4	<3	127	.78	.131	7	37	1.09	185	.11	3	3.54	.01	.13	<2	1.8
11250	1	71	5	83	.3	20	19	1904	4.41	11	<8	<2	2	43	<.2	3	<3	131	.94	.082	14	42	1.22	310	.12	4	3.41	.02	.14	<2	3.1
11300	<1	57	11	86	.4	21	16	904	4.24	8	<8	<2	<2	40	<.2	3	5	118	1.03	.058	12	36	.96	232	.12	4	3.14	.02	.17	<2	3.6
11350	<1	42	5	70	<.3	18	12	740	3.29	2	<8	<2	2	28	<.2	3	<3	99	.84	.050	9	35	.74	281	.15	3	2.48	.02	.18	<2	2.4
11400	<1	40	8	63	<.3	18	13	550	3.32	2	<8	<2	<2	37	.2	<3	<3	104	.65	.050	9	37	.80	167	.17	3	2.33	.02	.19	<2	3.1
11450	1	60	8	77	<.3	22	16	770	4.50	8	<8	<2	2	45	<.2	<3	<3	131	.80	.062	11	41	1.17	169	.16	4	3.02	.02	.11	<2	4.2
11500	1	48	9	84	<.3	18	14	1437	3.65	5	<8	<2	<2	37	.2	<3	4	105	.81	.060	9	31	.92	320	.12	4	2.69	.02	.11	<2	3.3
11550	<1	61	8	71	<.3	20	15	694	4.12	4	<8	<2	2	68	.2	<3	<3	117	1.27	.059	8	35	1.16	186	.11	<3	3.73	.02	.09	<2	2.7
11600	<1	42	3	92	<.3	15	17	1338	4.75	4	<8	<2	2	36	<.2	<3	<3	128	.86	.066	7	28	.89	264	.14	3	2.41	.02	.11	<2	4.2
11650	<1	56	3	59	<.3	17	13	786	3.78	5	<8	<2	<2	35	<.2	3	<3	114	.87	.078	8	33	.88	178	.14	4	2.30	.02	.11	<2	3.4
11700	1	32	4	67	<.3	13	12	931	3.31	5	<8	<2	2	33	.2	5	<3	96	.78	.044	8	24	.66	284	.14	5	2.45	.02	.17	<2	.8
RE 11700	1	33	4	68	<.3	13	12	969	3.35	4	<8	<2	<2	33	<.2	3	3	96	.78	.044	8	24	.67	290	.14	4	2.46	.02	.17	<2	1.0
11750	1	75	7	94	<.3	16	16	1109	5.26	5	<8	<2	2	34	.2	<3	4	122	.87	.052	10	30	1.02	280	.10	4	3.09	.02	.19	<2	3.3
11800	<1	50	7	77	<.3	19	14	688	3.90	5	<8	<2	2	43	.2	<3	<3	117	.77	.069	9	36	.99	140	.15	3	2.42	.02	.12	<2	5.1
11850	<1	44	9	72	<.3	17	13	613	3.93	2	<8	<2	2	40	<.2	<3	<3	120	.72	.056	8	33	1.10	133	.18	3	2.49	.01	.16	<2	1.1
11900	<1	59	7	86	.3	15	14	747	4.34	8	<8	<2	2	49	.2	4	<3	134	.94	.048	8	29	.98	144	.18	4	2.62	.02	.11	<2	2.4
11950	<1	46	7	65	<.3	13	12	630	3.69	<2	<8	<2	<2	34	<.2	<3	<3	116	.95	.032	9	25	.76	112	.17	4	2.35	.02	.11	<2	1.1
12000	2	29	8	72	<.3	9	13	929	3.25	5	<8	<2	<2	60	<.2	<3	<3	85	.84	.069	4	15	.62	246	.13	4	2.22	.02	.09	<2	.7
12000A	<1	63	6	61	<.3	15	15	592	4.24	5	<8	<2	<2	43	.3	4	<3	128	.81	.042	8	29	1.05	235	.17	3	2.61	.01	.08	<2	1.6
12600	1	33	5	69	<.3	13	12	788	3.28	3	<8	<2	<2	26	<.2	<3	<3	92	.54	.047	6	23	.71	154	.11	3	2.45	.02	.10	<2	2.2
12650	1	46	10	70	<.3	15	11	1043	3.56	11	<8	<2	2	40	<.2	<3	3	104	.93	.032	10	28	.76	177	.14	4	2.54	.02	.09	<2	2.3
12700	1	58	8	92	<.3	20	15	578	4.64	27	<8	<2	2	23	.2	4	3	115	.39	.222	6	27	1.05	165	.08	3	4.85	.01	.10	<2	2.5
12750	1	35	8	60	<.3	12	11	455	3.46	6	<8	<2	2	37	<.2	<3	<3	104	.81	.027	7	27	.70	114	.15	4	2.22	.02	.06	<2	2.0
12800	1	67	11	67	.3	18	18	695	5.62	15	<8	<2	2	42	.5	4	<3	125	.94	.046	9	29	.96	118	.16	3	2.68	.02	.09	<2	5.7
12850	8	74	22	114	.3	27	30	711	8.22	37	<8	<2	2	45	.7	4	<3	125	1.01	.053	9	28	1.03	116	.13	3	2.89	.02	.06	<2	4.2
12900	1	37	8	87	<.3	13	16	1335	3.98	3	<8	<2	<2	28	.3	<3	<3	95	.61	.069	8	24	.73	150	.10	<3	3.09	.01	.11	<2	1.2
12950	2	49	10	80	<.3	12	17	1085	4.23	17	<8	<2	<2	34	.2	<3	3	97	.69	.077	7	23	1.02	108	.11	<3	2.53	.01	.10	<2	4.1
1275050	<1	42	<3	58	<.3	6	13	1048	4.36	4	<8	<2	<2	74	<.2	3	<3	97	1.43	.053	7	10	1.28	72	.18	<3	3.74	.01	.14	<2	1.6
1275100	<1	49	<3	50	<.3	9	12	792	3.90	3	<8	<2	<2	86	.2	<3	<3	106	1.61	.054	5	17	.98	122	.15	<3	3.51	.02	.08	<2	2.7
STANDARD DS2	14	123	30	151	<.3	35	11	788	2.98	56	21	<2	4	28	9.9	12	11	73	.52	.086	16	161	.58	146	.09	5	1.70	.04	.16	7	206.0

GROUP 10 - 0.50 GM SAMPLE LEACHED WITH 3 ML 2-2-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR, DILUTED TO 10 ML, ANALYSED BY ICP-ES.
UPPER LIMITS - AG, AU, HG, W = 100 PPM; MO, CO, CD, SB, BI, TH, U & B = 2,000 PPM; CU, PB, ZN, NI, MN, AS, V, LA, CR = 10,000 PPM.
- SAMPLE TYPE: SOIL SS80 60C AU* BY ACID LEACHED, ANALYZE BY ICP-MS. (10 gm)
Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

DATE RECEIVED: MAR 28 2001 DATE REPORT MAILED: April 6/01 SIGNED BY: *[Signature]* D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS



SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm	Au* ppb
1275150	<1	51	3	62	<.3	12	15	766	4.23	<2	<8	<2	<2	110	<.2	3	<3	120	1.14	.056	6	23	1.25	146	.17	3	3.29	.01	.08	<2	2.1
1275200	<1	66	6	68	<.3	12	17	1005	5.03	<2	<8	<2	<2	123	.3	4	<3	152	1.59	.046	5	25	1.55	151	.23	4	4.05	.02	.10	<2	2.9
1275250	1	61	14	77	.3	16	21	1399	4.80	13	<8	<2	<2	69	.3	5	<3	121	1.49	.083	6	25	1.34	111	.17	4	2.54	.02	.07	<2	3.1
1275300	<1	34	10	78	<.3	13	15	912	3.54	5	<8	<2	<2	30	<.2	<3	<3	93	.68	.057	7	26	.72	139	.11	3	2.20	.01	.18	<2	1.3
1275350	<1	53	8	74	.3	13	15	994	4.06	3	<8	<2	<2	42	<.2	<3	<3	112	.95	.076	8	25	.95	166	.13	3	2.76	.01	.20	<2	1.9
1275400	<1	30	7	73	<.3	14	13	742	3.38	6	<8	<2	2	29	<.2	4	<3	97	.64	.050	5	24	.71	119	.14	4	1.93	.02	.16	<2	.9
1275450	<1	36	8	62	<.3	13	13	759	3.33	7	<8	<2	2	27	<.2	3	<3	91	.59	.054	6	26	.66	115	.12	4	1.96	.01	.27	<2	6.3
1275500	<1	30	5	59	<.3	13	12	550	3.44	5	<8	<2	<2	28	<.2	4	<3	102	.59	.060	4	30	.75	103	.13	3	1.89	.01	.14	<2	2.1
14000	<1	34	13	70	<.3	14	11	543	3.39	7	<8	<2	<2	28	<.2	4	<3	103	.61	.061	6	29	.81	99	.14	4	1.79	.01	.08	<2	4.0
14050	<1	38	15	89	<.3	15	12	588	3.62	6	<8	<2	2	36	<.2	5	<3	108	.68	.059	6	29	.93	123	.15	3	2.12	.02	.09	<2	2.7
14100	<1	56	20	101	<.3	16	13	665	3.99	8	<8	<2	2	41	<.2	<3	<3	121	.84	.061	8	32	.99	120	.15	<3	2.33	.02	.07	<2	5.6
14150	<1	43	29	129	<.3	18	14	731	3.91	7	<8	<2	2	42	<.2	3	<3	118	.74	.053	8	33	1.02	126	.16	4	2.46	.02	.13	<2	2.5
14200	<1	41	38	110	<.3	18	13	719	3.67	8	<8	<2	2	46	<.2	4	<3	113	.78	.046	8	33	.89	141	.14	4	2.49	.01	.10	<2	1.3
14250	<1	58	26	131	<.3	19	15	755	4.15	13	<8	<2	<2	51	.3	4	<3	122	.87	.055	10	34	1.12	137	.15	3	2.72	.01	.10	<2	3.0
14300	<1	50	27	116	<.3	20	16	814	4.24	13	<8	<2	2	51	.2	4	<3	127	.83	.050	10	37	1.10	128	.16	3	2.61	.02	.13	<2	2.2
14350	<1	71	20	112	.5	20	14	711	4.06	12	<8	<2	2	43	.2	3	<3	112	.99	.058	10	33	1.06	148	.13	4	2.68	.02	.13	<2	4.6
14400	<1	72	24	111	<.3	20	14	701	4.33	12	<8	<2	2	52	<.2	4	<3	116	.97	.074	10	35	1.15	125	.12	3	2.71	.02	.12	<2	3.7
14450	<1	57	25	161	<.3	16	12	905	2.89	4	<8	<2	2	42	.7	<3	<3	76	1.25	.040	9	24	.64	141	.10	6	2.38	.02	.09	<2	1.2
RE 14450	<1	60	27	171	<.3	16	12	951	2.98	6	<8	<2	<2	45	.8	3	<3	78	1.33	.043	9	24	.66	147	.11	7	2.52	.02	.10	<2	1.2
1400050	<1	24	21	127	<.3	12	11	760	2.80	5	<8	<2	2	26	.2	<3	<3	82	.55	.048	5	22	.60	122	.12	<3	2.07	.01	.09	<2	.5
1400100	<1	23	13	95	<.3	12	11	536	2.92	<2	<8	<2	2	26	<.2	3	<3	91	.58	.055	4	25	.63	116	.14	3	1.96	.01	.08	<2	1.0
1400150	<1	21	8	66	<.3	11	9	436	2.63	3	<8	<2	<2	34	<.2	<3	<3	83	.72	.028	5	23	.56	121	.14	4	1.94	.02	.06	<2	.6
1400200	1	38	16	79	<.3	14	12	613	3.51	8	<8	<2	2	48	.2	3	<3	106	1.00	.040	6	27	.84	135	.14	5	2.47	.02	.08	<2	1.6
1400250	<1	38	11	76	<.3	15	12	682	3.37	9	<8	<2	2	39	<.2	3	<3	99	.87	.043	7	30	.79	135	.14	4	2.20	.02	.11	<2	2.4
1400300	1	18	9	99	<.3	11	9	712	2.38	2	<8	<2	<2	26	<.2	3	<3	69	.60	.049	4	21	.47	128	.11	3	1.77	.02	.11	<2	4.2
1320050	1	41	13	78	<.3	18	15	829	3.57	7	<8	<2	2	40	<.2	<3	<3	107	.78	.052	8	34	.84	122	.14	4	2.41	.02	.13	<2	1.1
1320100	<1	40	15	80	<.3	19	15	837	3.60	<2	<8	<2	2	41	.2	<3	<3	107	.76	.046	9	33	.89	115	.15	4	2.34	.02	.12	<2	6.5
1320150	<1	30	10	81	<.3	15	13	639	3.26	<2	<8	<2	<2	38	<.2	<3	<3	103	.69	.042	7	32	.76	127	.15	<3	2.30	.02	.08	<2	2.1
1320200	1	58	16	71	<.3	19	19	955	5.92	10	<8	<2	2	49	<.2	6	<3	157	.89	.081	8	35	1.11	92	.18	<3	2.81	.02	.08	<2	3.6
13000	<1	31	8	75	<.3	14	15	1054	3.38	<2	<8	<2	2	32	<.2	3	<3	98	.59	.049	7	28	.70	135	.12	<3	2.62	.01	.10	<2	.9
1300050	<1	52	9	67	<.3	18	12	650	4.02	3	<8	<2	2	51	<.2	<3	<3	117	1.19	.071	9	32	1.08	117	.15	5	2.55	.03	.10	<2	3.6
1300100	<1	56	8	57	<.3	18	13	587	4.34	7	<8	<2	2	45	<.2	3	<3	121	1.02	.061	8	36	1.01	123	.15	4	2.56	.02	.12	<2	3.0
1300150	<1	38	6	53	<.3	14	10	444	3.88	<2	<8	<2	2	59	<.2	<3	<3	108	1.47	.065	7	25	1.00	113	.16	3	3.07	.17	.14	<2	3.2
STANDARD DS2	14	124	35	153	.3	35	11	796	2.99	56	17	<2	5	28	10.1	11	9	73	.52	.087	17	164	.59	146	.09	6	1.71	.04	.16	7	202.6

Sample type: SOIL SS80 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.



SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm	Au* ppb
1300200	1	30	8	68	<.3	15	13	624	3.19	4	<8	<2	2	29	<.2	<3	3	91	.64	.035	7	26	.66	124	.12	3	2.09	.02	.12	<2	1.5
1300250	1	28	8	74	<.3	13	13	745	3.10	3	<8	<2	<2	27	<.2	<3	<3	88	.58	.043	6	23	.67	125	.11	3	1.89	.01	.10	<2	1.2
1300300	1	32	9	75	<.3	15	14	782	3.45	3	<8	<2	<2	31	<.2	3	<3	97	.65	.051	6	26	.77	136	.12	3	2.09	.01	.13	<2	1.3
1300350	1	34	11	70	<.3	14	15	847	3.52	2	<8	<2	<2	33	<.2	<3	<3	99	.66	.044	8	26	.77	131	.12	3	2.17	.02	.08	<2	1.9
13150	1	34	10	70	<.3	16	14	675	3.76	4	<8	<2	2	35	<.2	<3	<3	105	.59	.058	7	30	.78	130	.14	3	2.45	.01	.10	<2	1.4
13200	1	38	15	88	<.3	17	14	767	3.46	4	<8	<2	<2	41	<.2	3	<3	105	.73	.054	8	34	.83	124	.14	4	2.18	.01	.12	<2	1.2
13250	1	37	12	85	<.3	16	14	869	3.27	<2	<8	<2	<2	33	.3	<3	<3	97	.69	.053	7	29	.81	109	.13	4	2.02	.01	.13	<2	1.6
13300	<1	50	14	94	<.3	18	16	921	3.62	4	<8	<2	2	42	.3	<3	<3	104	1.06	.058	8	31	.94	115	.14	3	2.25	.02	.11	<2	4.0
13350	<1	67	12	88	<.3	22	18	1010	4.36	10	<8	<2	2	46	<.2	<3	<3	125	.94	.074	9	41	1.19	124	.15	4	2.69	.02	.13	<2	3.9
13400	1	65	42	154	<.3	20	17	944	4.29	12	<8	<2	2	39	.3	7	<3	121	.92	.082	8	35	1.30	124	.15	4	2.66	.02	.13	<2	6.9
RE 13400	<1	60	37	145	<.3	20	16	895	4.05	9	<8	<2	2	37	.3	5	<3	113	.87	.079	7	33	1.23	117	.14	4	2.49	.02	.12	<2	3.2
13450	<1	63	16	125	<.3	20	16	983	4.19	5	<8	<2	2	36	.2	<3	<3	109	1.07	.058	10	36	1.08	235	.15	4	3.42	.02	.20	<2	2.0
13500	<1	35	15	103	<.3	17	17	889	3.89	7	<8	<2	<2	28	.2	3	<3	110	.56	.059	6	28	1.00	100	.13	4	2.80	.01	.07	<2	1.4
13550	1	49	29	129	<.3	22	18	1051	4.42	13	<8	<2	2	34	.2	3	<3	124	.75	.069	6	35	1.19	152	.13	4	3.61	.01	.10	<2	3.0
13600	1	76	35	143	<.3	20	20	1154	5.17	14	<8	<2	2	58	.4	<3	<3	142	1.06	.071	9	35	1.30	129	.16	4	3.23	.02	.08	<2	10.4
13650	1	59	9	73	<.3	16	14	913	3.98	12	<8	<2	2	49	<.2	4	<3	117	.95	.086	7	35	.98	88	.16	4	2.26	.03	.07	<2	10.9
13700	1	74	16	103	<.3	22	21	1224	4.58	14	<8	<2	2	63	.3	<3	<3	129	2.80	.088	7	36	1.37	136	.15	5	2.59	.03	.10	<2	3.9
13750	1	51	10	123	<.3	7	16	1187	2.89	10	<8	<2	<2	38	.3	<3	<3	69	.92	.172	4	13	.59	115	.06	4	2.44	.01	.11	<2	3.4
13800	1	56	13	121	<.3	14	15	765	3.68	6	<8	<2	<2	53	.2	3	<3	112	.67	.075	8	30	.89	200	.15	3	2.49	.01	.06	<2	3.1
13850	1	53	18	109	<.3	17	14	616	3.82	4	<8	<2	2	38	<.2	<3	<3	111	.69	.061	9	27	.82	159	.13	5	3.01	.01	.15	<2	3.2
13900	<1	38	27	117	<.3	15	14	603	3.83	7	<8	<2	2	33	.3	<3	<3	122	.69	.055	7	31	.88	159	.17	5	2.25	.02	.15	<2	3.1
13950	<1	64	39	115	<.3	16	12	587	4.20	8	<8	<2	2	33	<.2	3	<3	123	.88	.051	8	31	.91	153	.15	4	2.65	.03	.13	<2	4.4
13770	1	2400	359	4773	3.1	4	19	2222	5.78	53	<8	<2	<2	25	28.9	47	4	109	4.97	.078	6	6	1.52	27	.03	5	2.29	.01	.05	<2	671.7
STANDARD DS2	14	125	33	152	<.3	35	11	797	2.97	55	21	<2	4	28	9.9	9	11	72	.51	.086	17	161	.58	146	.09	7	1.69	.04	.16	7	195.8

Sample type: SOIL SS80 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.



GEOCHEMICAL ANALYSIS CERTIFICATE



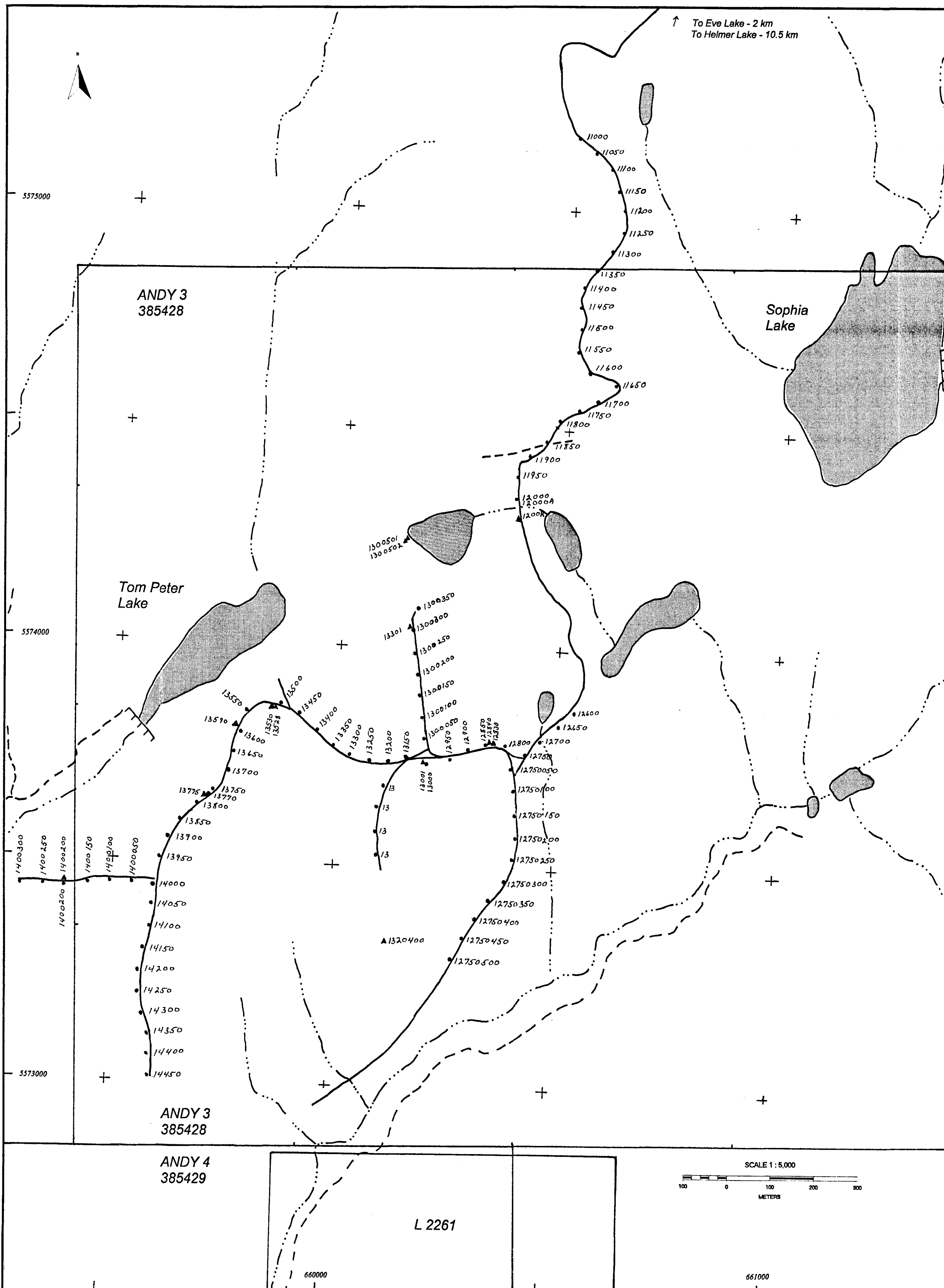
Hayes, T. File # A100862

1704 - 555 Austin Ave, Coquitlam BC V3K 6R8 Submitted by: Ted Hayes

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm	Au* ppb
12000R	2	10	3	33	<.3	6	14	840	5.46	10	<8	<2	<2	40	<.2	3	<3	93	1.12	.115	4	9	1.55	48	.21	7	2.76	.16	.14	<2	1.9
12830	6	55	<3	49	<.3	5	15	636	5.81	34	<8	<2	<2	31	<.2	<3	<3	182	.61	.080	2	13	1.79	34	.21	5	1.99	.14	.20	<2	2.4
12840	3	48	<3	59	<.3	4	11	844	4.78	14	<8	<2	<2	24	.4	<3	<3	141	1.74	.066	2	7	1.44	24	.19	8	2.46	.16	.09	<2	3.4
13001	2	41	<3	39	<.3	2	5	281	6.12	10	<8	<2	<2	37	<.2	<3	<3	119	.76	.060	2	9	.47	109	.27	<3	1.86	.03	.18	<2	.4
13528	1	7	10	61	<.3	11	16	997	4.93	16	<8	<2	<2	58	.6	4	<3	45	7.35	.051	3	3	1.24	79	<.01	19	.69	.02	.16	<2	1.3
13530	3	21	<3	157	<.3	7	17	1189	5.78	13	<8	<2	<2	22	1.7	5	<3	151	2.55	.109	4	16	2.16	15	.33	94	2.61	.12	.03	<2	2.0
13590	5	58	<3	107	<.3	12	16	988	5.38	11	<8	<2	<2	31	1.3	4	<3	153	3.42	.122	6	17	1.64	33	.24	10	2.76	.12	.07	<2	2.4
13775	5	1751	26	3015	1.5	6	4	584	1.95	8	<8	<2	<2	2	17.3	8	<3	21	.09	.016	1	24	.36	15	.01	4	.66	.01	.08	<2	2.4
1300301	3	46	27	197	.8	12	28	2014	6.71	51	<8	<2	<2	18	.2	<3	<3	141	.40	.070	4	21	1.71	87	.04	8	2.52	.07	.24	<2	3.6
1305001	2	47	3	56	.7	17	35	732	5.74	14	<8	<2	<2	13	.9	<3	<3	171	3.07	.042	<1	25	1.94	35	.13	9	3.79	.09	.07	<2	2.2
1305002	2	45	8	50	1.3	19	36	827	8.23	49	<8	<2	<2	13	.2	3	3	211	.59	.054	1	34	2.43	50	.20	3	2.37	.11	.14	<2	4.1
RE 1305002	2	45	6	50	1.3	17	36	823	8.14	46	<8	<2	<2	13	<.2	<3	<3	208	.58	.053	<1	33	2.41	48	.19	3	2.32	.10	.13	<2	3.5
1320400	2	48	<3	83	<.3	14	26	1678	7.28	9	<8	<2	<2	28	.5	<3	<3	225	2.02	.071	3	29	2.66	35	.42	9	3.90	.09	.06	<2	1.5
1400200	2	134	<3	94	<.3	4	20	1445	7.32	8	<8	<2	2	22	.4	4	<3	210	2.17	.110	5	12	2.48	69	.50	11	3.26	.06	.02	<2	1.5
STANDARD C3/DS2	26	61	29	158	5.5	34	11	754	3.39	55	20	<2	21	29	21.6	16	21	83	.56	.092	18	169	.58	150	.09	21	1.80	.04	.16	16	214.1
STANDARD G2	1	3	3	42	<.3	8	4	534	2.06	<2	<8	<2	4	76	<.2	<3	<3	43	.65	.102	7	80	.58	233	.13	4	.97	.09	.49	2	-

GROUP 1D - 0.50 GM SAMPLE LEACHED WITH 3 ML 2-2-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR, DILUTED TO 10 ML, ANALYSED BY ICP-ES.
 UPPER LIMITS - AG, AU, HG, W = 100 PPM; MO, CO, CD, SB, BI, TH, U & B = 2,000 PPM; CU, PB, ZN, NI, MN, AS, V, LA, CR = 10,000 PPM.
 ASSAY RECOMMENDED FOR ROCK AND CORE SAMPLES IF CU PB ZN AS > 1%, AG > 30 PPM & AU > 1000 PPB
 - SAMPLE TYPE: ROCK R150 60C AU* BY ACID LEACHED, ANALYZE BY ICP-MS. (10 gm)
 Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

DATE RECEIVED: MAR 28 2001 DATE REPORT MAILED: April 6/01 SIGNED BY: *[Signature]* D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS



GEOLOGICAL SURVEY BRANCH
ASSESSMENT REPORT

26,872

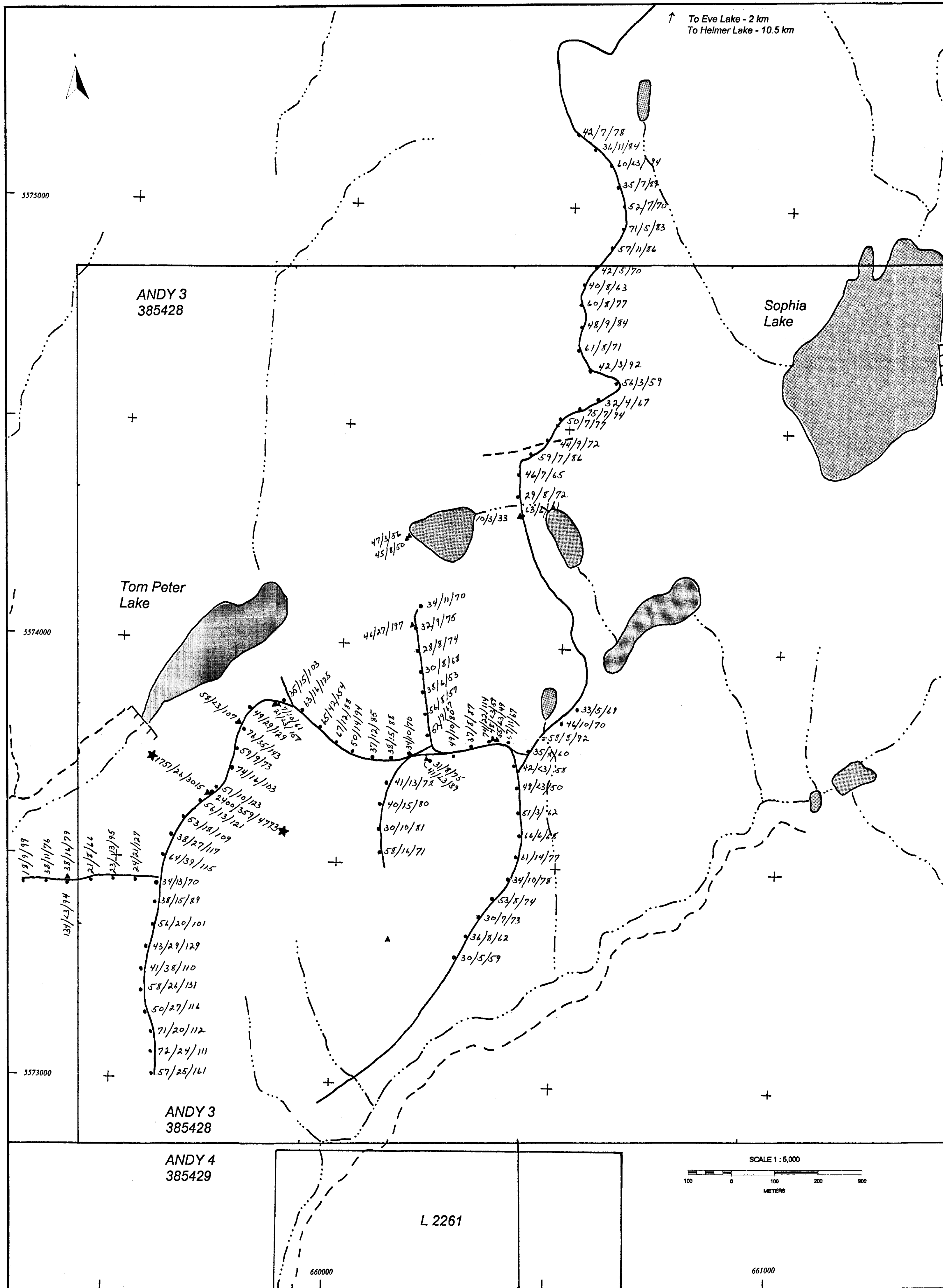
- LEGEND**
- Lake & Creek
 - Old Gravel Road
 - Logging Road
 - Soil Sample Location, Sample Number
 - Rock Sample Location, Sample Number
 - Lease
 - Claim Line

Note: Claim Lines were positioned by chain & compass in relation to topography
Sample locations were positioned with a GARMIN 12XL hand-held GPS

AHURA MINING LTD.
ANDY PROJECT
Nicola Mining Division, B.C.

SOIL AND ROCK GEOCHEMISTRY
SAMPLE LOCATION ①

Project No:	By:	M.A.
Scale:	Drawn by:	M.A.
Date:	Figure:	5



SOIL SAMPLES				
ELEMENT SAMPLES	Cu ppm	Pb ppm	Zn ppm	Au ¹ ppb
11000	42	7	78	3.0
11050	38	11	84	2.4
11100	60	<3	94	2.0
11150	35	7	89	1.4
11200	52	7	70	1.8
11250	71	5	83	3.1
11300	57	11	86	3.6
11350	42	5	70	2.4
11400	40	8	63	3.1
11450	60	8	77	4.2
11500	48	9	84	3.3
11550	61	8	71	2.7
11600	42	3	92	4.2
11650	56	3	59	3.4
11700	32	4	87	0.8
11750	75	7	94	3.3
11800	50	7	77	5.1
11850	44	9	72	1.1
11900	59	7	88	2.4
11950	46	7	65	1.1
12000	29	8	72	0.7
12000A	63	6	61	1.6
12600	33	5	69	2.2
12650	46	10	70	2.3
12700	58	8	92	2.5
12750	35	8	60	2.0
12800	67	11	67	5.7
12850	74	22	114	4.2
12900	37	8	87	1.2
12950	49	10	80	4.1
1275050	42	<3	58	1.6
1275100	49	<3	50	2.7
1275150	51	3	62	2.1
1275200	68	6	68	2.9
1275250	61	14	77	3.1
1275300	34	10	78	1.3
1275350	53	8	74	1.9
1275400	30	7	73	0.9
1275450	36	8	82	6.3
1275500	30	5	59	2.1
14000	34	13	70	4.0
14050	38	15	89	2.7
14100	56	20	101	5.6
14150	43	29	129	2.5
14200	41	38	110	1.3
14250	58	28	131	3.0
14300	50	27	118	2.2
14350	71	20	112	4.6
14400	72	24	111	3.7
14450	57	25	181	1.2
1400050	24	21	127	0.5
1400100	23	13	85	1.0
1400150	21	8	68	0.8
1400200	38	16	79	1.6
1400250	38	11	76	2.4
1400300	18	9	99	4.2
1320050	41	13	78	1.1
1320100	40	15	80	6.5
1320150	30	10	81	2.1
1320200	58	16	71	3.6
13000	31	8	75	0.9
1300050	52	9	67	3.6
1300100	56	8	57	3.0
1300150	38	6	53	3.2
1300200	30	8	68	1.5
1300250	28	8	74	1.2
1300300	32	9	75	1.3
1300350	34	11	70	1.9
13150	34	10	70	1.4
13200	38	15	88	1.2
13250	37	12	85	1.6
13300	50	14	94	4.0
13350	67	12	88	3.9
13400	65	42	154	6.9
13450	63	16	125	2.0
13500	35	15	103	1.4
13550	49	29	129	3.0
13600	76	35	143	10.4
13650	59	9	73	10.9
13700	74	16	103	3.9
13750	51	10	123	3.4
13800	56	13	121	3.1
13850	53	18	109	3.2
13900	38	27	117	3.1
13950	64	39	115	4.4
13770	2400	359	4773	671.7

ROCK SAMPLES				
ELEMENT SAMPLES	Cu ppm	Pb ppm	Zn ppm	Au ¹ ppb
12000R	10	3	33	1.9
12630	55	<3	49	2.4
12840	48	<3	59	3.4
13001	41	<3	39	0.4
13526	7	10	61	1.3
13530	21	<3	157	2.0
13590	58	<3	107	2.4
13775	1751	26	3015	2.4
1300301	46	27	197	6.6
1305001	47	3	56	2.2
1305002	45	8	50	4.1
1320400	48	<3	83	1.5
1400200	134	<3	94	1.5

LEGEND

- Lake & Creek
- Old Gravel Road
- Logging Road
- 42/7/78 Soil Sample Location, Cu / Pb / Zn
- 10/3/33 Rock Sample Location, Cu / Pb / Zn
- L2261 Lease
- Claim Line

Note: Claim Lines were positioned by chain & compass in relation to topography
Sample locations were positioned with a GARMIN 12XL hand-held GPS

AHURA MINING LTD.
ANDY PROJECT
Nicola Mining Division, B.C.

**SOIL AND ROCK GEOCHEMISTRY
SAMPLE RESULTS**
Copper, Lead, Zinc (2)

Project No:	1:5,000	By:	M.A.
Scale:	May-02	Drawn by:	M.A.
Date:		Figure:	6

26,872

GEOLOGICAL SURVEY BRANCH
ASSESSMENT