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LIST OF MAPS

- FIG 1. LOCATION MAP
- FIG 2. CLAIM MAP
- FIG 3. SAMPLE LOCATION MAP

ATTACHMENTS

1. ANALYTICAL RESULTS

GEOLOGICAL BRANCH ASSESSMENT REPORT

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1. INTRODUCTION

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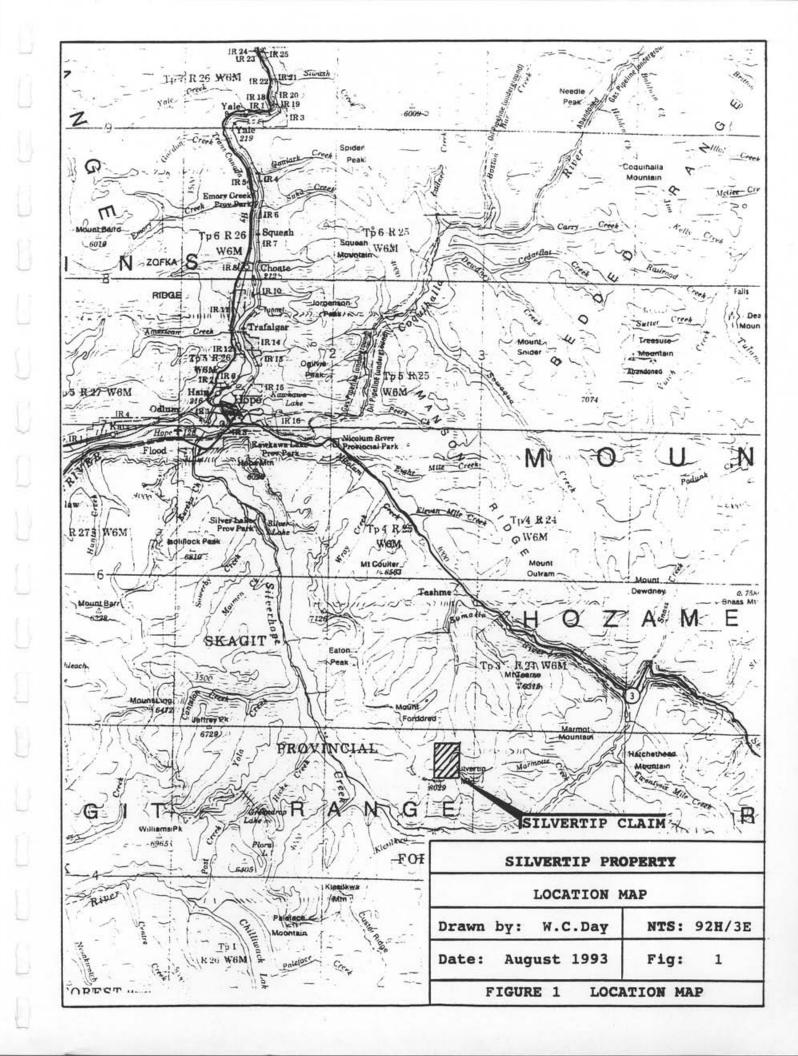
- 1.1 This report has been prepared at the request of Mr. A.E. Angus who is the registered owner of the Silvertip Claim.
- 1.2 The Silvertip claim consists of 12 claim units which are located near the headwaters of the Sumallo River in Southwestern British Columbia.
- 1.3 At present the commodities of interest, in order of importance, are zinc, copper, lead, silver (gold).

2. SUMMARY

- 2.1 A short property assessment was conducted on the Silvertip claim between July 4 and July 10, 1993. Geochemical sampling was conducted and general prospecting undertaken during the period.
- 2.2 The samples collected during the program were analyzed by Van Geochem Labs Ltd., of Vancouver B.C. The results of these analyses indicate the presence of significant zinc and subordinant copper and lead mineralization within the claim area. A phased program consisting of geological mapping, geochemical sampling and diamond drilling is recommended to further evaluate the property.

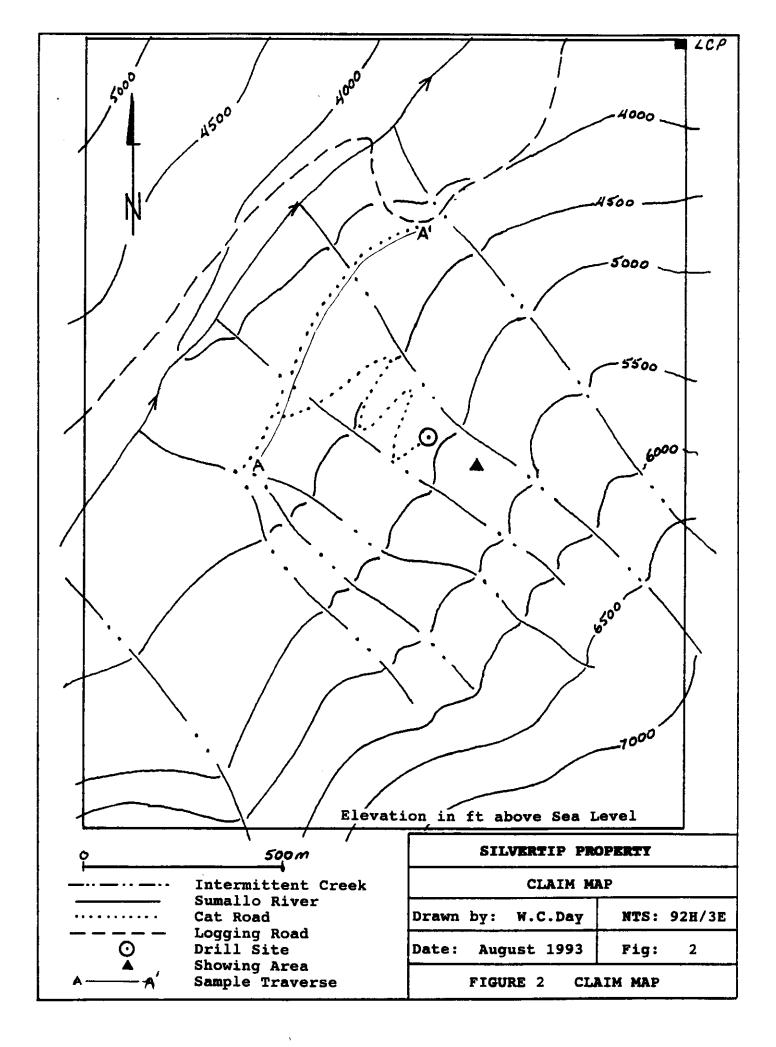
3. LOCATION & ACCESS

- 3.1 The Silvertip claim is located some 27 km southeast of Hope, B.C. near the headwaters of the Sumallo River on the northwest flank of Silvertip Mountain.
- 3.2 The claim area can be gained by utilizing a logging road from Sunshine Village (formerly Tashme) which is located on Highway 3 S.E. of Hope and then travelling 13 km south.
- 3.3 The claim is located in an area in which the terrane could be described as steep to precipitous. Evidence of avalanche and rockslide activity is widespread. The property rises from a low of 1125 meters to a high of 2300 meters above sea level.
- 3.4 Second growth consisting of alder and minor conifers cover the areas that were accessible to logging. Areas inaccessible to logging are largely barren outcrop or talus slopes. Mature fir stands are found in the balance of the inaccessible areas.
- 3.5 A historical drill station is located near the center of the claim and can be accessed on foot by using a washed out and rubble littered cat trail which was constructed during the mid '60's.



4. CLAIM DATA

- 4.1 The Silvertip claim consists of 12 claim units with the legal corner past located at the northeast corner. From this site the claim area extends three claim units west and four claim units south. Its record number is 311389 and the expiry date is July 11/96.
- 4.2 The registered owner of the claim is Mr. A.E. Angus of Surrey, B.C.
- 4.3 The claim is located in the New Westminster Mining District and is bounded within NTS 92H/3E and centered at approximately 49° 10'N latitude and 121° 15' W longitude.



5. HISTORY

- 5.1 Though undocumented, mineralization was apparently located within the Silvertip claim area prior to 1965 as a diamond drill program was conducted between the period Nov 10/65 to March 25/66. Two holes were drilled with the 1st drilled at N 63° E dipping at -14° for a hole length of 1329'. The 2nd was drilled at N 76° E and dipping at -10° to a hole length of 562'. A steep tractor trail was also constructed to the drill site but apparently was unused.
- 5.2 The drill was supplied by helicopter and water was hoisted to the site by cable and sled from the valley floor. The drill site was at 4900' (1494m) above sea level. The property at this time was owned by Allison Pass Mining Ltd. of Vancouver.

5.3 A report for Allison Pass Mining Ltd. cites the presence of significant mineralized zones being encountered in both holes drilled. For example: HOLE 1A Sample # Length Cut ₽b% Znt Aq opt 8 ' 21 .05 1.18 .93 .6 HOLE 2A Sample # Length Cu% Pb% Znt Ag opt 100' 13 .07 .28 .07 N/A

Apparently a spectrographic analytical technique was used. Surface showings are apparently also present, though not witnessed during the current program. These showings are said to have been traced for 600 feet along surface and to be open at both ends. The width is said to be not fully exposed, however, a 16' width sampled (#17) is said to average 3.42% zinc, .1% lead and .08% copper. No drill logs or full analytical results were

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attached to this report.

5.4 The only subsequent evaluation of the claim area was conducted by Suecon Development Corp. The claim area was visited by a representative of this company on July 18/83 and on August 27, 1984.

6. GEOLOGY

- 6.1 The Silvertip claim is underlain by rocks of the Hozameen Group (Monger 1970 and McTaggart & Thompson 1967). This Group consists of four divisions: ribbon chert, basic lavas (greenstone), limestone and argillite.
- 6.2 During late Palaeozoic or Triassic time these rocks were metamorphosed to form the Custer Gneiss. A layered high grade migmatitic complex of gneiss and schist. Later high grade metamorphism during late Cretaceous time is associated with the injection of the Spuzzum Intrusions followed by the foliated granodiorite Yale Intrusions and deposition of Eocene conglomerate and sandstone and the Intrusion of the Chilliwack batholithic rocks. The Custer Gneiss is separated from the overlying Hozameen rocks by a fault zone and the Hozameen fault separates the Hozameen beds from Mesozoic formations to the east.
- Plutonism of Miocene to Oligocene age was accompanied by 6.3 extensive vulcanism and mineralization in the Cascade A number of important base and precious metal Range. deposits in B.C. and Washington State are associated with this activity. One of these, the Canam deposit, lies 16 km to the east and occurs in а breccia pipe. Mineralization in this deposit is copper with lesser amounts of gold, silver, tungsten and uranium.

subject in this report consist of dominantly greenstone, lesser chert, tuff and minor argillite. Significant mineralization consisted dominantly of zinc with lesser copper and lead and, as indicated in the analytical results, silver and very minor gold.

6.5 The mineralization occurs in three modes - layered (bedded?), fracture filling (breccia healing), and disseminated. these modes are typified by the rock samples numbered 48653, 48655 and 48656. A description of each of these samples follows.

Sample No. Description 48653 Tuff with 2% magnetite as disseminations, fracture fillings and blebs, minor chalcopyrite and pyrite.

- 48655 Layered cherty argillite with pyrrhotite, sphalerite, chalcopyrite and minor magnetite along bedding? planes.
- 48656 Brecciated tuff healed with pyrrhotite, magnetite and minor chalcopyrite

7. SUBJECT PROGRAM

- 7.1 The program was conducted during the period July 4 and July 10 1993 and was initially to include geological mapping, geochemical sampling (rock, soil and silt) and prospecting. The extreme terrane conditions however, precluded any attempt at geological mapping at this time. Several attempts were made to gain access to the mineral showing cited to be present some 100 m above the drill station but each was unsuccessful.
- 7.2 It was decided that, due to the short duration of the program, selective sampling of rocks in the talus and a soil sample traverse below the suspected zone of mineralization would assist in the preliminary assessment of the claim. To this end a soil sample traverse was conducted. Soil samples (51) were collected at 20 meter intervals from above the cut of the lower cat track. Selected rock samples (14) from talus were collected and tied into this traverse. Ten silt samples were also collected during the period.

8. RESULTS

- 8.1 Analyses of the samples collected (14 rock, 51 soils and 10 silts) was conducted by Van Geochem Labs of Vancouver, B.C. The soil and silt samples were subjected to multielement analyses by ICP. Rock samples were subjected to multi-element analyses by ICP and fire assayed with a atomic absorption finish to assess gold content.
- 8.2 Five rock samples exceeded the detection limit for zinc (2%) and one exceeded the limit for lead (2%). All rock samples were found to have anomalous copper values, three were anomalous in lead and all were anomalous to highly anomalous in zinc. Each of the soil and silt samples were anomalous to highly anomalous in zinc. 39 of the 51 soil samples was anomalous in lead and all but one anomalous in copper. Anomalous values are considered to be those in excess of 100 ppm in each case. The fire assay results show one rock sample (48654) to be highly anomalous in gold (200 ppb) and several others to be elevated in that element (20 - 40 ppb). Several of the rock samples were also anomalous in silver (plus 1 ppm) with one sample (48660) having a very significant value of greater than 50 ppm.
- 8.3 There has been no horizonal development of the soil which is essentially rock dust resulting from pulverization of the up slope rocks during landslide/avalanche activity. As a result the base metal content of these soils is considered to be directly associated with a mineralized zone present up slope of the traverse line. The width/thickness of this zone is unknown, however its length is indicated to be potentially substantial as the entire traverse (1000 m) is enriched in zinc, and particularly that area between samples ST 15 and ST 47

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(640 m). This same area shows the greatest enrichment of copper, lead and silver.

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9. CONCLUSIONS AND RECOMMENDATIONS

- 9.1 The results of the program indicate that a zone of base metal enrichment of unknown dimensions is present on the Silvertip claim. Zinc appears to be the principal commodity present with subordinate copper and lead.
- 9.2 The terrane in the claim area is very rugged and limits accessibility by foot. As a result a helicopter assisted program is recommended to explore the claim more fully. Geological mapping and rock chip sampling could be accomplished by having a helicopter lift personnel into specific areas for mapping and sampling. It should also remain available to move personnel when conditions dictate. Interpersonal and helicopter communication by walkie talkie would be a great asset and the use of ropes both for safety reasons and for greater mobility in some areas would assist the program.
- 9.3 A two stage program is recommended. Stage one would incorporate a program as lined out in 9.2. To conduct the program, two experienced field personnel should be used. A time period of two weeks should be allotted and scheduled when weather conditions would be most favourable for flying (summer, early fall). Stage two would entail diamond drilling.

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10. PROPOSED BUDGET STAGE I

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Personnel
2 @ $400.00/day 14 days
                                                       $11,200
     (2 days mob/demob, 3 days down due to weather,
     9 days field)
Vehicle
                                                       $ 1,200
                                                       $ 2,100
Room & Board
                                                       $ 2,500
Analyses
Disposables
                                                       $ 1,000
Report
                                                       $ 1,200
Helicopter 2 hrs/day, 9 days @ $600/hr
                                                      $ 10,800
     Total
                                                      $ 30,000
```

The stage 2 program would undoubtedly constitute a drill program. No budget is forwarded at this time however, as site locations, footage etc. are fully contingent upon the results of stage I. Any drill program envisioned would require full helicopter support.



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11. STATEMENT OF COSTS - JULY 93 PROGRAM

Personnel	
Geologist	\$1,200.00
Prospector	9 00. 00
Food	21 9.4 2
Gas	102.95
Camp @ \$25/day/man	3 00. 00
Vehicle 4x4 pickup @ \$60/day	3 60. 00
Trail Bike \$50/day	3 00. 00
Report & Preparation	7 00. 00
Assaying	<u>744.</u> 14

Total

\$4,826.51

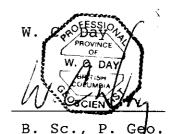


STATEMENT OF QUALIFICATIONS

I, William Colin Day, residing @ 2409 Western Ave, North Vancouver, B.C. hereby certify that:

- I was a member of the crew which conducted the program of subject in this report.
- I am a graduate of the University of British Columbia (B.Sc. Geology 1976).
- 3. I have been practising my profession as an exploration geologist since 1976.
- 4. I am a member in good standing of the Association of Professional Engineers and Geoscientists of B.C.
- 5. I hold no interest in the Silvertip claims, nor do I expect to receive any.

August 3, 1993 North Vancouver, B.C.



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ANALYTICAL RESULTS

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ICAP GEOCHEMICAL ANALYSIS

A .5 gram sample is digested with 5 ml of 3:1:2 HCL to HMO₃ to H₂O at 95 °C for 90 minutes and is diluted to 10 ml with water. This leach is partial for Al, **Da**, Ca, Cr, Fe, K, Hg, Hn, Ha, P, Sn, Sr and W.

ANALYST: 21game

REPORT 8: 930059 PA	BILL DAY					PROJEC	T: NONE	GIVEN			DATE	IN: JULI	14 1993	DATE	OUT: JU	ILY 21 19	193 AT	TENTION:	NR. BILI					D PAGE 1	OF 2
Sample Name	Ag pp a	Al X	As p pa	Ba p pa	Bi p pn	Ca X	Cd p pe	Co p pa	Cr 998	Cu ppe	Fe I	K I	Mg X	fin p pe	No p pa	Na Z	Ni p pa	Р 1	Pb ppa	Sb p ga	Sn p pa	Sr 998	U ppa	¥ pp e	Zn p pe
ST-1	0.5	4.53	(3	45	3	1.32	2.3	70	138	138	7.77	<0.01	3.59	1797	3	0.01	125	0.11	14	(2	(2	38	(5	<3	606
\$1-2	0.3	4.51	<3	39	(3	2.00	(0.1	75	132	135	7.74	<0.01	3.58	1758	(1	<0.01	126	0.12	19	<2	<2	43	(5	(3	550
ST-3	0.3	4.64	(3	47	(3	1.33	0.6	78	139	168	8.15	<0.01	3.54	2082	1	0.02	135	0.12	28	<2	(2	41	(5	<3	681
ST-4	0.2	4.94	(3	43	(3	1.31	0.5	74	142	144	8.18	(0.01	3.76	1866	(1	0.01	125	0.12	10	(2	(2	37	(5	(3	658
\$1-5	0.3	4.79	<3	46	(3	1.44	0.2	11	137	162	8.28	<0.01	3.62	1883	4	0.01	128	0.12	16	<2	<2	43	(5	<3	697
ST-6	0.6	5.02	(3	49	(3	1.53	0.1	84	138	164	8.43	(0.01	3.78	2139	(1	0.01	138	0.12	28	(2	(2	43	(5	(3	663
ST-7	0.3	5.05	<3	49	(3	1.45	(0.1	90	133	186	8.56	<0.01	3.68	2188	(1	0.01	145	0.11	33	<2	(2	45	(5	(3	702
ST-8	0.3	4.44	(3	49	(3	1.50	0.9	79	134	164	8.18	(0.01	3.42	2018	2	0.01	132	0.13	31	(2	(2	44	(5	(3	686
ST-9	0.5	5.01	<3	48	<3	1.47	0.7	68	146	180	8.82	<0.01	3.68	2116	2	0.03	148	0.12	40	(2	(2	55	(5	(3	979
ST-10	0.1	5.03	(3	44	(3	1.30	1.5	78	143	164	8.21	<0.01	3.77	2180	(1	0.01	135	0.10	24	<2	<2	45	<5	(3	844
ST-11 ST-12	0.5	4.79	(3	49	(3	1.44	2.7	88	145	207	8.60	<0.01	3.58	2369	4	0.02	162	0.12	51	(2	<2 (2	53	(5 /5	(3 , (3	991 978
	0.6	4.76	(3	50	(3	1.41	2.3	86 74	140	205	8.86	(0.01	3.43	2255	1	0.04	150	0.13	39	<2	<2	53 52	(5	(3	
ST-13 ST-14	0.7	4.64	(3 (3	50 48	(3	1.41	2.8	74	133	188	8.44	<0.01	3.41	2007	3	0.04	137	0.14	95	(2	<2	34	<5 /5	(3	942 851
ST-15	1.0	4.33	(3	70 56	(3	1.15	3.0	69	146	161	7.83	(0.01	3.59	2163	1	<0.01	142	0.14	252	<2	(2	66	(5		
21-17	1.5	4.30	(3	30	(3	2.93	3.0	70	155	177	7.86	<0.01	3.60	2183	2	<0.01	152	0.13	297	<2	<2	66	<5	<3	1252
ST-16	1.5	4.34	<3	53	<3	1.17	3.3	71	155	177	7.98	(0.01	3.64	2247	1	<0.01	157	0.14	329	<2	(2 ·	35	(5	<3	1184
ST-17	1.5	4.34	<3	54	<3	3.04	5.2	73	161	178	7.98	(0.01	3.61	2219	2	<0.01	151	0.12	237	<2	<2	71	<5	<3	1440
ST-18	1.1	4.34	(3	56	(3	2.97	3.9	73	159	178	7.94	<0.01	3.63	2175	1	<0.01	155	0.12	246	<2	<2	70	(5	<3	1330
ST-19	1.2	4.46	<3	53	(3	2.92	5.5	70	165	179	7.98	<0.01	3.65	2214	1	<0.01	149	0.12	233	<2	(2	71	(5	(3	1593
ST-20	1.0	4.18	<3	47	(3	3.91	4.7	62	153	149	7.43	{0.01	3.52	2023	1	(0.01	133	0.11	245	<2	<2	84	<5	<3	1226
ST-21	1.1	4.40	(3	50	<3	4.14	5.9	68	163	181	7.94	<0.01	3.64	2151	2	(0.01	145	0.12	218	<2	(2	93	<5	(3	1534
ST-22	1.1	4.54	(3	54	(3	4.15	5.2	67	160	176	7.90	(0.01	3.68	2125	(1	(0.01	145	0.12	218	(2	<2	94	<5	(3	1523
SI-23	0.8	4.76	(3	39	(3	1.37	3.5	69	160	157	7.92	(0.01	3.66	2202	1	(0.01	145	0.12	109	(2	<2	37	<5	(3	991
ST-24	1.2	4.29	(3	55	(3	1.00	2.9	75	170	163	8.19	(0.01	3.57	2663	6	(0.01	164	0.14	135	<2	(2	23	(5	<3	1512
ST-25	1.2	4.65	(3	48	(3	1.21	5.1	72	179	167	8.33	<0.01	3.74	2491	4	<0.01	169	0.12	173	<2	(2	38	<5	(3	1269
ST-26	1.1	4.55	<3	52	<3	1.14	3.2	68	176	149	7.87	<0.01	3.83	2317	i	(0.01	163	0.13	214	{2	<2	30	۲۵	<3	1291
SI-27	1.5	4.35	(3	57	<3	1.66	2.3	65	167	157	7.79	(0.01	3.70	2265	2	(0.01	161	0.13	381	<2	<2	41	<5	(3	1085
SI-28	0.8	4.59	(3	59	(3	1.13	3.5	74	174	175	8.33	(0.01	3.79	2448	1	(0.01	162	0.13	177	<2	(2	29	<5	(3	1312
ST-29	1.0	3.98	(3	53	(3	1.34	2.5	62	157	142	7.29	(0.01	3.47	2234	2	(0.01	150	0.13	300	(2	(2	35	(5	(3	1043
ST-30	0.6	3.98	(3	39	<3	1.01	1.2	55	155	85	7.40	(0.01	3.15	2315	3	(0.01	118	0.12	111	<2	<2	25	(5	{3	947
SI-31	0.5	4.32	٢3	58	(3	1.06	7.7	70	175	168	8.35	(0.Ů1	3.67	2494	2	<0.01	168	0.12	124	(2	<2	27	<5	<3	2068
SI-32	1.0	4.51	(3	62	(3	1.22	8.3	72	182	170	8.54	<0.01	3.78	2560	2	(0.01	175	0.13	131	(2	<2	31	<5	<3	1980
ST-33	0.9	4.60	(3	55	(3	1.01	1.1	71	172	168	8.45	(0.01	3.79	2486	<1	(0.01	167	0.12	126	<2	<2	26	<5	<3	1913
ST-34	1.1	4.48	(3	67	(3	1.34	1.1	71	177	167	8.54	(0.01	3.80	2524	4	(0.01	170	0.13	123	<2	(2	33	<5	(3	1930
ST-35	ů. 4	2.32	<i>4</i> 3	45	(3	1.69	7.3	58	152	134	6.63	(C.01	2.91	2095	\$	(0.01	133	0.10	125	(2	(2	42	~S	3	1471
ST-36	ŋ.7	4.26	(3	E4	<3	2.51	٤.5	££	:69	148	8.11	(0.01	3.67	2307	3	(0.01	158	0.12	126	(2	٢2	56	(5	<3	1617
SI- 37	Ù. 9	4.18	<3	57	(3	1.37	6.9	66	162	129	7.93	(0.01	2.51	2925	3	(0.01	150	0.14	141	<2	<2	32	<5	₹3	1605
ST-38	0.9	4.45	<3	53	(3	ú.95	5.4	70	170	130	8.49	(0.01	3.71	2665	3	(0.01	155	0.13	142	(2	(2	21	<5	₹3	1540
51-39	0,8	4.32	<3	60	(3	1.31	13.7	75	15E	150	8.04	(0.01	3.57	2751	4	(0.01	165	0.13	126	<2	<2	31	<5	<3	2512
Minique Detection	<u>0.1</u>	6.01	3	1	3	0.01	0.1	1	1	1	ė.0:	0.0i	0.6;	1	I	0.01	1	0.01	2	2	2	:	5	3	1
Maxioum Detection	50.0	10,00	2000	1000	1000		1690.0	20060	1060	20000	10.0ù	10.00	10.0¢	20000	1000	10.00	20000	10.00	2000 0	2000	1000	10000	100	1000	20000
C - Less Than Minimum) - Greater	Than Maxi	B UA	is - Ins	sufficies	it Sample	ê As	- No S3a	ple	ANOMALO	IS RESUL	TS - Fui	ther Anal	VSES B	Alteine	ta Metho	ds Succe	sted.							

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ICAP GEOCHEMICAL ANALYSIS

A .5 gram sample is digested with 5 ml of 3:1:2 HCL to HMOs to HgO at 95 °C for 90 minutes and is diluted to 10 ml with water. This leach is partial for Al, Ba, Ca, Cr, Fe, K, Mg, Mn, Na, P, Sn, Sr and W.

REPORT 8: 930059 PA	BILL DAY					PROJE	CT: NONE	GIVEN			DATE	IN: JUL	14 199:	3 DATE	E OUT: JU	RLY 21 19	993 . A1	TENTION:	NR. BILI	DAY				PAGE 2	OF 2
Sample Name	Ag	AL	As	Ba	Bi ppa	Ca	Cd	Co ppa	Cr	Cu	fe	ĸ	Hg	Xn ppo	No p pe	Na 7	Ni ppa	P	Pb ppa	Sb p pe	Sa p pa	Sr ppa	U ppa	Н рр п	Zn ppe
ST-40	рр е 1.2	4.06	рра (3	ppa 56	 	1.17	ppe 8.2	69	рра 149	рра 156	7.87	<0.01	3.41	2506	4	<0.01	152	0.12	146	<2	<2	33	(5	(3	1680
ST-41	1.0	4.24	(3	56	(3	1.15	7.7	69	168	158	8.36	(0.01	3.71	2475	1	(0.01	160	0.12	138	<2	(2	28	(5	(3	1710
51-42	2.7	4.32	(3	55	(3	1.10	7.1	70	165	151	8.47	<0.01	3.74	2667	3	(0.0)	157	0.14	130	(2	(2	27	(5	(3	1650
ST-43							6.5						3.72				149	0.13	126	(2	(2	25	(5	(3	1592
81-13 ST-44	1.0 1.0	4.29 4.01	<3 <3	52 55	(3 (3	1.04	6.9	68 66	168 153	152 150	8.26 7.82	(0.01	3.45	2526 2436		<0.01 <0.01	145	0.12	134	(2	(2	27	(5	(3	1574
31-14	1.0	4.01	(3	23	(3	1.12	8.7	00	133	130	1.62	<0.01	3.43	2930	1	(0.01	143	V. 12	134	12	14		13	(3	13/4
ST-45	1.2	4.15	(3	53	(3	1.10	7.1	70	161	158	8.24	(0.01	3.63	2564	3	<0.01	151	0.12	133	<2	<2	27	<5	(3	1739
SI-46	1.0	3.98	(3	54	(3	1.24	5.7	64	148	141	7.53	<0.01	3.43	2415	(1	(0.01	146	0.13	131	<2	<2	33	<5	<3	1588
ST-47	1.5	4.02	(3	81	(3	1.02	4.6	76	145	168	8.22	<0.01	3.37	2975	6	(0.01	155	0.13	185	<2	(2	30	(5	<3	1518
ST-48	1.5	3.37	(3	11	(3	0.81	3.2	65	92	150	7.43	(0.01	2.47	3290	6	(0.01	114	0.15	312	<2	<2	25	(5	(3	816
ST-49	1.9	3.52	(3	70	<3	0.60	(0.1	63	95	159	7.68	<0.01	2.50	2953	6	<0.01	111	0.15	307	<2	<2	19	<5	(3	745
ST-50	1.8	3.45	<3	83	<3	0.65	(0.1	69	91	163	7.86	<0.01	2.44	3591	8	<0.01	120	0.15	342	<2	<2	21	<5	⟨3	799
ST-51	2.1	3.49	(3	87	(3	0.57	(0.1	72	90	180	8.04	(6.01	2.48	3641	9	(0.01	126	0.14	339	(2	(2	21	(5	(3	821
STC-1	0.6	2.59	(3	72	(3	0.80	(0.1	38	83	87	5.42	(0.01	2.13	1497	1	(0.01	78	0.12	12	(2	(2	25	(5	<3	227
STC-2	0.6	2.43	(3	47	(3	0.77	(0.1	35	82	68	5.62	(0.01	2.07	1246	i	(0.01	75	0.11	13	(2	(2	22	(5	(3	212
S1C-3	0.2	2.86	<3	141	(3	0.99	(0.1	33	83	73	5.03	<0.01	1.41	1496	1	(0.01	79	0.11	(2	<2	<2	32	<5	<3	164
STC-4	0.1	2.77	(3	134	(3	0.84	(0.1	31	83	64	4.99	<0.01	1.48	1326	' 1	<0.01	81	0.09	(2	<2	(2	27	<5	(3	143
STC-5	0.1	2.33	(3	187	(3	0.72	(0.1	29	60	98	4.53	(0.01	1.16	1472	1	(0.01	70	0.11	<2	<2	<2	27	(5	<3	170
STC-6	0.1	2.18	(3	161	(3	0.73	(0.1	24	56	74	4.05	(0.01	1.14	1258	i	(0.01	56	0.09	<2	(2	(2	26	(5	(3	174
STC-7	0.1	2.30	(3	201	(3	0.49	(0.1	40	65	100	5.30	(0.01	1.51	1756	i	(0.01	94	0.10	4	(2	(2	20	(5	(3	237
SIC-B	0.1	2.28	<3	211	(3	0.52	(0.1	40	65	102	5.25	<0.01	1.47	1785	i	(0.01	101	0.10	8	(2	(2	21	(5	(3	234
S1C-9	0.2	3.07	<3	72	(3	0.95	(0.1	40	104	76	5.91	(0.01	2.74	1211	(1	<0.01	90	0.09	<2	<2	(2	20	<5	<3	145
SIC-10	0.2	3.11	<3	112	(3	1.00	(0.1	46	107	86	6.34	(0.01	2.70	1413	1	(0.01	97	0.10	<2	<2 <2	(2	27	(5	<3	171
Minimum Detection	0.1	0.01	3	1	3	0.01	0.1	1	1	1	0.01	0.01	0.01	1	1	0.01	1	0.01	2	2	2	1	5	3	1
Naxious Detection	50.0	10.00	2000	1000	1000	10.00	1000.0	20000	1000	20000	10.00	10.00	10.00	20000	1000	10.00	20000	10.00	20000	2000	1000	10000	100	1000	20000
< - Less Than Minisum	> - Greater				sufficien			- No Sam						lyses By										••••	
	/ 0160761					·· Acahi	- 43	NU 388	4.2	- AUIMLU	SA NEADE	10 101		1353 61		e nesuu	as andde	#*EU:							

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ICAP GEOCHEMICAL ANALYSIS

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A .5 gram sample is digested with 5 ml of 3:1:2 HCL to HNO₂ to H₂O at 95 °C for 90 minutes and is diluted to 10 ml with water. This leach is partial for Al, Ba, Ca, Cr, Fe, K, Mg, Mn, Na, P, Sn, Sr and N.

				Α.	5 grae s	ample is			al of 3: Sis parti									o 10 el	with val	ier .		ANALY	'ST: _	<u></u>	am	al
REPORT #: 930060 PA	BIL	I, DAY					PROJE	CT: NONE	GIVEN			DATE	IN: JULY	14 1993	DATE	OUT: JU	LY 21 19	1A E9	TENTION	: MR. BILI	L DAY				PAGE 1	OF 1
Sample Name	Ag	A1	As	+Au	Ba	Bi	Ca	Cd	Ĉo	Cr	Cu	Fe	K	Ħg	Mn	No	Na	Ni	P	Pb	Sb	Sn	Sr	U	H	In
	pge	1	ppe	ppb	ppa	ppe	1	pp	p p a	ppe	ppe	Ĩ	1	1	pon	ppe	I	ppn	I	ppe	ppe	ppe	pps	ppe	pps	ppa
48651	0.7	4.11	(3	20	9	(3	3.63	71.2	111	125	552)10	(0.01	1.70	1232	16	0.37	64	0.04	(2	(2	<2 <2	85	(5 (5	(3 (3	9677 270
48632	0.5	3.99	(3	10	a	<3	5.34	(0.1	111	148	553	>10	(0.01	1.65	1010	15	0.32	11	0.05	(2	(2	-	120	-		
48653	រាទ	05	NS.	ĥs	ns	15	05	ns.	A\$	85	AS	05	R5	NS	85 3150	85	N5	NS	RS	NS (2	115 (2	ns (2	ns 83	ns (5	ns (3	ns)200 00
48654	3.0	3.31	(3	200	17	31	3.80	364.1	128	89	581 866	>10	(0.01	1.50 1.79	3159	19	1.24	36 68	0.04 0.08	(2 26	<2 <2	(2	22	(5	(3	18782
48655	0.9	2.06	(3	10	(1	(3	1.98	133.6	198	44	110D	>10	(0.01	1.73	1615	18	0.15	00	Ų.VG	20	1	12	22	13	14	10/02
48656	N.5	65	ns	រាទ	រាទ	NS	ns	AS	65	ns	пs	ns	85	ns	រាទ	N 5	85	ns	AS	ñs	กร	AS	ns	กร	A 5	ns
48657	1.2	1.52	(3	10	(1	(3	0.82	317.4	102	42	546	>10	<0.01	1.30	1196	12	0.4B	65	0,05	<2	<2	<2	9	<5)20000
48658	1.9	2.02	<3	30	(1	(3	8.94	253.2	231	47	644	>10	(0.01	1.77	8110	16	0.79	38	0.13	- 14	<2	(2	39	<5	(3	>20000
48659	0.5	4.68	<3	30	(1	(3	4.75	1.4	254	98	501	>10	(0.01	1.36	1294	19	0.31	BO	Q. 19	<2	<2	(2	170	<5	(3	897
48660	>50	0.24	<3	20	35	(3	0.32	92.1	12	61	355	5.01	(0.01	0.13	492	5	0.10	12	0.04	>20000	80	(2	13	(5	(3	18776
48661	2.2	0.59	(3	30	(1	⟨3	0.26	241.2	32	50	338	>10	(0.01	0.64	1091	12	0.36	15	0.01	506	<2	<2	3	(5	(3	>20000
48662	0.7	2.84	<3	40	4	3	2.74	35.4	62	148	263	8.78	(0.01	1.18	1035	9	0.21	78	0.04	194	(2	(2	81	<5	, (3	4930
48663	1.1	3.17	<3	20	()	<3	1.75	238.0	16	59	462	>10	(0.01	3.37	2986	8	0.25	33	0.03	35	<2	<2	8	(5	(3	>20000
48664	0.4	1.10	<3	10	15	(3	1.17	90.7	12	69	149	5.64	(0.01	1.09	898	12	0.01	32	0.08	37	<2	<2	12	(5	<3	13464
Кіпзаца Detection Махівца Detection К — Less Than Kinibum	0.1 50.0 > - 1	0.01 10.00 Greater	3 2000 Than Naxi	5 10000	1 1000 is - Ins	3 1000 sufficier	0.01 10.00 t Sampl	0.1 1000.0 e ns	1 20000 - No Sam	1 1000 ple	1 20000 4Au Anai	0.61 10.00 Iysis Do	0.01 1 0.00 ne By Fi	0.01 10.00 re Assay	1 20000 Concentr	5 1000 ation /	0.01 10.00 AAS Fin:	1 20000 ish.	0.01 10.00	2 20000	2 2000	2 1000	1 1 00 00	5 100	1000 3	1 20000

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ICAP GEOCHEMICAL ANALYSIS

A .S gram sample is digested with 5 ml of 3:1:2 HCL to HNO₂ to H₂O at 95 °C for 90 minutes and is diluted to 10 ml with water. This leach is partial for Al, Ba, Ca, Cr, Fe, K, Mg, Mn, Wa, P, Sn, Sr and W.

				Α.	,) gram s	ampie 15			a ni of a h is part									ið 1V H i	ALCU AS	ier.		ANAL	/ST: _	Ľ.	zam	al
REPORT 1: 930063 PA	NR.	BILL DA	NY				PROJE	ct: NONE	GIVEN			DATE	IN: JUC	¥ 21 1993	DATE	. OUT: JU	JLY 26 1	1A 20	FTENT LON	: MR. BIL	L DAY				PAGE 1	OF 1
Sample Name	Ag	AL	As	4Au	Ba	Bi	Ca	Cđ	Co	Cr	Cu	fe	ĸ	Ng	Mn	No	Ka	Ni	P	РЪ	Sb	Sn	Sr	U	W	ln
	ppe	1	ppe	ppb	pps	ppe	ĩ	pps	ppa	ppm	ppa	1	1	1	ppa	ppe	1	ppa	1	ppe	ppa	opa	ppe	ppe	pp=	pp=
48653	1.2	1.60	(3	30		(3	0.48	<0.L	258	98	852	>10	(0.01	1.51	986	18	0.03	165	0.02	30	(2	<2	19	(5	<3	101
48656	1.3	1.05	(3	30	a	(3	3.56	(0.1	281	102	763	>10	(0.0 1	1.00	886	20	0.05	171	0.03	37	<2	(2	47	<5	(3	182
Minimum Detection	0.1	.0.01	3	5	1	3	0.01	0.1	I.	1	1	0.01	0.01	0.01	1	1	0.01	1	0.01	2	2	2	1	5	3	1
Naxioun Detection (- Less Than Minioun	50.0 >-(10.00 Greater	2000 Than Maxi	10000 inua	1000 is - Ins	1000 Sufficier	l0.00 ht Sampl	1000.0 e ns	2000 0 - No Sa	100 0 mple	2000 0 #Au Anai	10.00 Lysis Bo	10.00 ne Dy Fi	10.00 re Assay	20000 Coaceatr	LOOQ ation /	10.00 MAS Fin	20000 ish.	10.00	20000	2000	1000	10000	[00	1000	20000

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VGC VANGEOCHEM LAB LIMITED

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MAIN OFFICE 1630 PANDORA STREET VANCOUVER, B.C. V5L 1L6 TEL (604) 251-5656 FAX (604) 254-5717

BRANCH OFFICES BATHURST, N.B. RENO, NEVADA, U.S.A.

REPORT NUMBER: \$30866 GA	JOB NUMBER: \$39060	HR. BILL DAY	PAGE 1 OF 1
SAMPLE #	Au		
	р рb		
48651	20		
48652	10		
48653			
48654	200		
48655	10		
48656	. 		
48657	10		
48658	30		
48659	30		
48660	20		
48661	30		
48662	40		
48663	20		

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VGC VANGEOCHEM LAB LIMITED

MAIN OFFICE 1630 PANDORA STREET VANCOUVER, B.C. V5L 1L6 TEL (604) 251-5656 FAX (604) 254-5717

BRANCH OFFICES BATHURST, N.B. RENO, NEVADA, U.S.A.

REPORT NUMBER: \$30063 GA	JOB NUNBER: \$30043	WR. BILL DAY	PAGE 1 OF 1
SAMPLE #	Au թ թb		
48653	30		
48656	30		

	Rock Sample #	Znppm	Cuppm	Pbppm	Agppm	1													
	48651 48652 48653 -48654 -48655	9677 270 101 +20000 18782	552 553 852 581 866	<2 <2 30 <2 26	0.7 0.5 1.2 3.0			s	ilt Sample	e # Zn	ppm	Сиррт	Suma Pbppm	allo Rive Agppm			STC 2	\sim	
N/	48656 48657 48658 48659 48660	182 +20000 +20000 897 18776	763 546 644 501 355	37 <2 14 <2 +20000	1.3 1.2 1.9 0.5 +50.0				STC 1 STC 2		227 212	87 68	12 13	0.6			5	STC 1	
48651	48661 48662 48663 48664	+20000 4930 +20000 13464	338 263 462 149	506 194 35 37	2.2 0.7 1.1 0.4														*
48652 48653 48654 48655 48657 48659 48657 48659	48660	48661	• •		• •	486	6,2 '	48663	48664	ST 4	10	• ST	F 45	• •	ST 50			TO SUNST	1jn
ST 1 ST 5 ST 10 ST 15	ST 2	20 Sample#	ST Znppm	25 Cuppm		T 30 Agppm	Sample#	Znppm	Cuppm	Pbppm	Agppm	Sample#	Znppm	Cuppm	Pbppm	Agppm			
	To Drill Sit	ST 1 ST 2 ST 3 ST 4 ST 5	606 550 681 658 697	138 135 168 144 162	14 19 28 10 16	0.5 0.3 0.2 0.3	' ST 21 ST 22 ST 23 ST 24 ST 25	1534 1523 991 1512 1269	181 176 157 163 167	218 218 109 135 173	1.1 1.1 0.8 1.2 1.2	ST 41 ST 42 ST 43 ST 44 ST 45	1710 1650 1592 1574 1739	158 151 152 150 158	138 · 130 126 134 133	1.0 2.7 1.0 1.0 1.2			
0 20 40 60 80 100 m	Sit.	ST 6 ST 7 ST 8 ST 9 ST 10	663 702 686 979 844	164 186 164 180 164	28 33 31 40 24	0.6 0.3 0.3 0.5 0.1	ST 26 ST 27 ST 28 ST 29 ST 30	1291 1085 1312 1048 947	149 157 175 142 185	214 381 177 300 111	1.1 1.5 0.8 1.0 0.6	ST 46 ST 47 ST 48 ST 49 ST 50	1588 1518 816 745 799	141 168 150 159 163	131 185 312 307 342	1.0 1.5 1.5 1.9 1.8			
	-	ST 11 ST 12 ST 13 ST 14	991 978 942 851	207 205 188 161	51 39 95 252	0.5 0.6 0.7 1.0	ST 31 ST 32 ST 33 ST 34	2068 1980 1913 1930 1471	168 170 168 167 134	124 131 126 123 125	$0.9 \\ 1.0 \\ 0.9 \\ 1.1 \\ 0.9$	ST 51	821	180	339	2.1		IP PROPE	
		ST 15 ST 16 ST 17 ST 18 ST 19	1252 1184 1440 1330 1593	177 177 178 178 179	297 329 237 246 233	1.5 1.5 1.1 1.2	ST 35 ST 36 ST 37 ST 38 ST 39	1617 1605 1540 2512	148 129 130 150	126 141 142 126	O.7 0.9 0.9 0.8						Drawn by: W.C Date: August FIGURE 3 SA		NT Fi
		ST 20	1226	149	245	1.0	ST 40	1680	156	146	1.2						FIGURE 3 SA		AT.

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