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REPORT ON 1989 WORK
on
THE SNOWFIELD GOLD ZONE
and
THE SNOWFIELD GOSSAN AREA
SULPHURETS PROPERTY
SKEENA MINING DIVISION
N.T.S. 104-B/9

March 8, 1990

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SUMMARY

Taking advantage of exceptional snow melt conditions, a short exploration program was conducted around the **Snowfield Gold Zone** located on the north part of **Newhawk-Granduc's Sulphurets Property** north of **Stewart, B.C.** This work was funded and performed by **Corona** on behalf of the joint venture.

A picketed grid was installed and lithogeochemical sampling, prospecting and cursory mapping were carried out, and core from two 1968 Granduc drill holes was resampled.

The **Snowfield Gold Zone** was discovered in 1980 and has been estimated to contain 7.7 million tons of 0.08 opt Au in unimpressive looking chloritic andesite tuffs. 1989 results indicate that drilling in the south part of the zone should expand this resource significantly.

Prospecting discovered a number of new showings, the most significant of which was a 7.3 g/t Au sample from a silicified fault zone on the west end of the grid.

Future work should consist of extending the grid north, east and west for control purposes and to continue the lithogeochemical sampling to discern metal zoning patterns based on a porphyry model. Further geologic work should be detailed and concerted, as obtaining a sound geologic understanding of the area is essential to discovering further mineralization.

INTRODUCTION

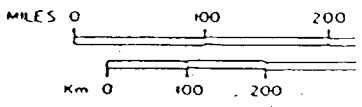
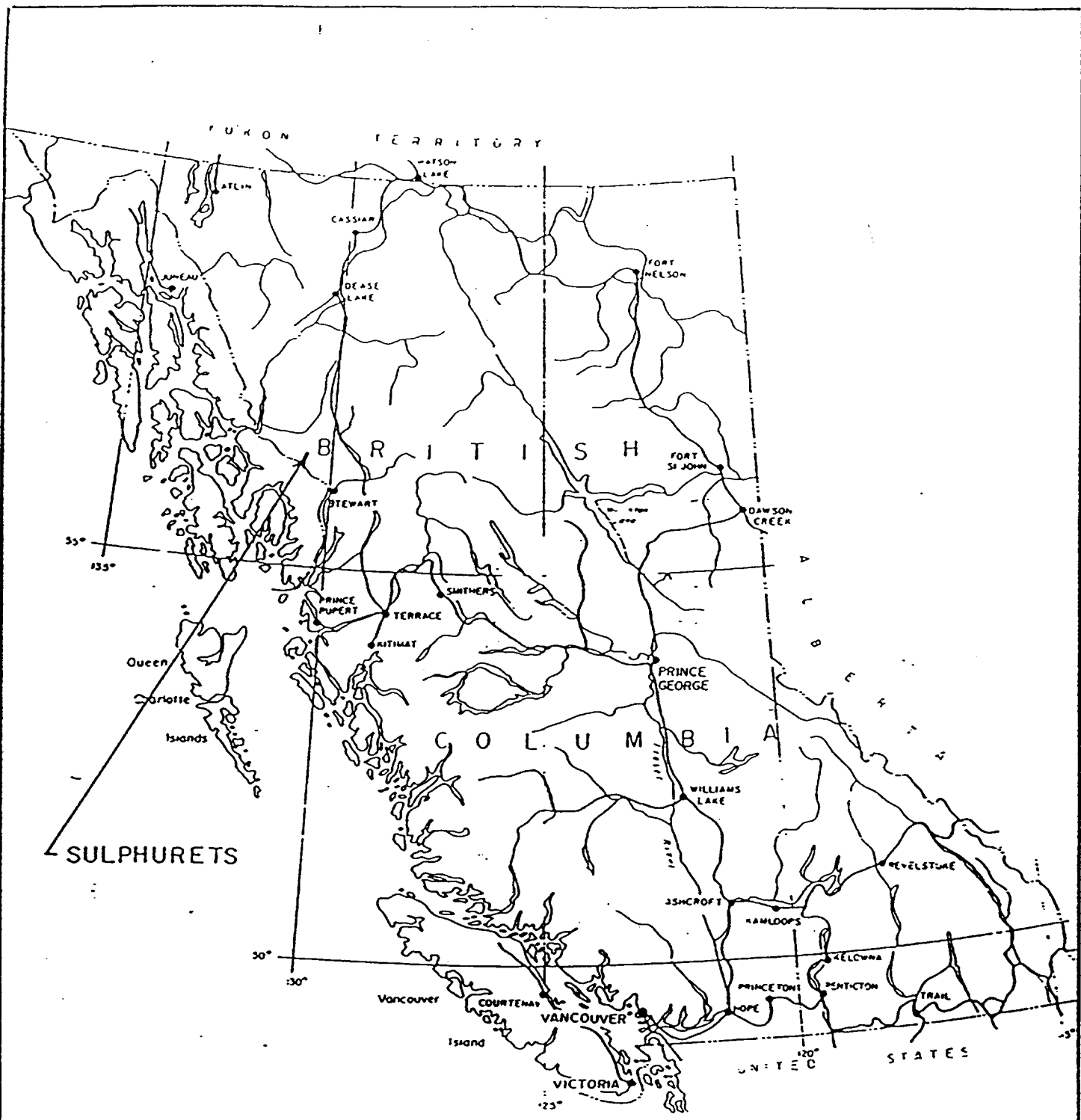
The Snowfield Gold Zone is located 5 km northwest of the Brucejack Au-Ag deposit on Newhawk Gold Mines-Granduc Mines' Sulphurets Property, 60 km north of Stewart, B.C. in the Coast Mountains of northwest B.C. The zone is situated on the north side of the Mitchell Sulphurets Ridge on the east end of the Snowfield Gossan at an elevation of 1590 m.

Snow conditions in late summer 1989 resulted in the best outcrop exposure in recent history and an excellent opportunity to further explore the Snowfield Gold Zone which had previously been extensively snow covered. The work was carried out from a fly camp on the gossan by R. Johnston, B. Laird and D. Gaunt from September 11 - 27, 1989. The project was fully funded by Corona Corporation from its 1989 Sulphurets Monitoring Budget.

HISTORY

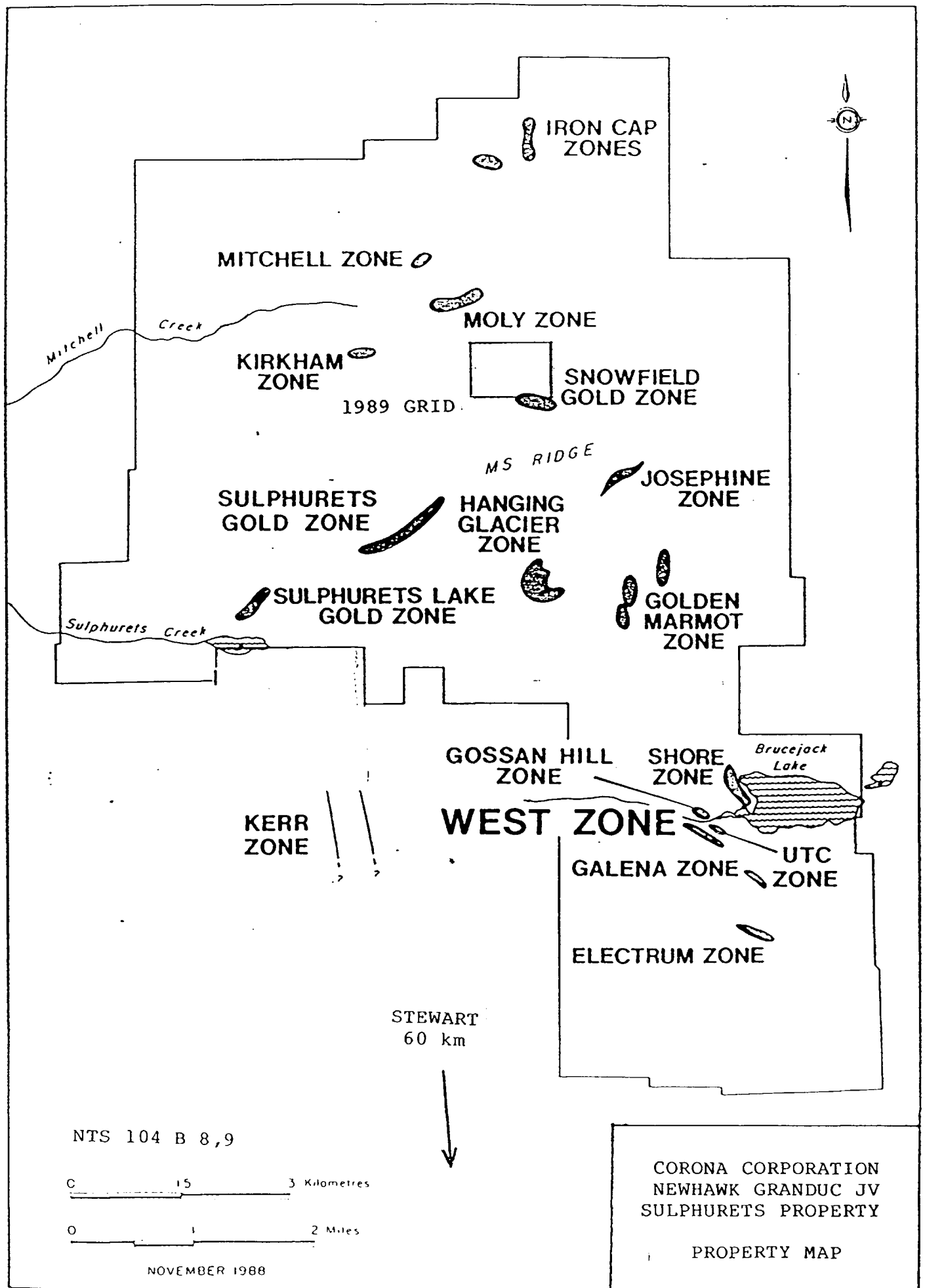
Prospecting and minor placer operations have been carried out in the Stewart-Unuk River area throughout this century, and Mitchell Creek has long been known to contain placer gold. Granduc Mines Ltd. staked the present Sulphurets Property in 1960 and subsequent work discovered areas of molybdenite mineralization on the north side of Mitchell-Sulphurets Ridge. The Quartz Stockwork Zone, located 200 m north of the Snowfield Gold Zone, was tested in 1968 by two 1,000 foot diamond^{*} drill holes, but only minor molybdenite mineralization was discovered.

Esso Minerals Canada optioned the property in 1980 and the Snowfield Gold Zone was discovered in that year as a result of a regional lithogeochemical sampling program. A 200 x 200 m area of chloritic andesite tuffs was found to contain consistent high Au values (>2000 ppb) and trenching carried out in 1981 outlined a large area grading >.04 opt Au.



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 NEWHAWK GRANDUC JV
 SULPHURETS PROPERTY

LOCATION MAP



In 1985 *Newhawk Gold Mines* drilled five diamond drill holes totalling 738.1 m into the auriferous area and outlined a resource of 7.7 million tons averaging 0.08 opt Au. Only cursory visits have been made to the property since 1985.

REGIONAL GEOLOGY

The *Mitchell Glacier* area is underlain by Jurassic sedimentary and volcanic rocks of the *Hazleton Group* which have been locally intruded by intermediate and felsic intrusives and later mafic dykes. Structural deformation has been extensive and large areas have been moderately to intensely sericitized, pyritized and oxidized.

1989 PROGRAM

A picketed grid was installed as a control for the mapping and lithogeochemical sampling program and should remain usable for a number of years. The grid is centred on the *Snowfield Gold Zone*; the collar of 85-112-114 is 0+00N/0+00W and the baseline extends from 1+50E to 10+00W across the *Snowfield Gossan*. A total of 8.025 km of grid lines were put in using 000° as grid north.

Rock samples were taken at each of the stations, and the area was prospected and further sampled, and a 1:2,500 geologic map was prepared. Core from the two *Granduc* drill holes in the *Quartz Stockwork Zone* was logged and resampled. As all of the core from both holes had previously been split, the remaining core was sent out for analysis, with a "skeleton" (3-4 pieces of core from each assay interval) retained at the drill collars.

LITHOLOGY AND ALTERATION

The *Snowfield Gossan* is a zone of intense sericite-pyrite ± quartz alteration; an overprint on a regional chloritization (propylitization?) of the *Mitchell-Sulphurets Ridge*. Most of the original textures have been obscured or obliterated, but the original rock type appears to be andesite lapilli tuff. Bedding measurements near 5+50W/2+00S indicate the tuffs to be overturned.

A small area of intensely chloritized intermediate intrusive was noted on two small nunataks at the south end of the grid around 3+75W/5+50S. The only other intrusives noted were a number of dark weathering fine grained mafic dykes in the south central part of the grid. Barite veins are spatially related to some of those.

Sericitization is intense throughout much of the gossan with pyrite content ranging from 2-20%. Silicification is locally present both as pervasive alteration and as veins which often contain minor sulphides.

STRUCTURE

The Snowfield Gossan is cut off on the east side by the Brucejack fault and to the west by a northwest fault referred to as the Ortum Fault. The gossan itself is cut by numerous regional and local shears which have altered most of the rocks into sericite schists. The more intensely sheared zones contain boudins and rods of quartz and local massive knots of pyrite.

Northwest faults which dip 30-55° northeast are the most major in the area. The area west of the Ortum Fault is made up of chloritized andesites cut by intensely sericitized shears, which is interpreted to be a high level representation of an intense zone of sericitization at depth (the Snowfield Gossan). This is similar to the scenario postulated for the Old Yeller Zone near Brucejack Lake indicating considerable vertical movement on these northwest faults.

North-south faults cut the northwest ones and have dips ranging from vertical to 65°E, and subvertical east-west faults are also present. Displacement on these faults is unknown.

LITHOGEOCHEMISTRY

As the **Snowfield Gold Zone** is visually indistinguishable from the surrounding chloritized tuffs, a lithogeochemical sampling program was undertaken as a tool to discover further such zones. Any attempts to contour the geochemical results should take into account two separate populations of samples; more or less systematically collected samples from grid stations; and prospecting samples which tend to be high grade grabs and not necessarily representative of the location.

Most of the grid area is extremely high in Au with background values in the >100 ppb range. The **Snowfield Gold Zone** is easily recognized as a 300 x 350 m area in which Au values are consistently >1000 ppb and up to 4060 ppb, surrounded by a 200-300 m wide halo of 300+ ppb Au. No other **Snowfield** type areas were discovered in the sampling program.

The **Coffee Pot Zone** shows up as a distinct 200-300 ppb Au anomaly.

The eastern half of the grid is strongly anomalous in Mo with values >50 ppm, forming a wide halo around the **Snowfield Gold Zone**, while in the western half of the grid Mo content is less than 20 ppm. This break coincides with an interpreted north-south fault at about 4+50W.

Zinc anomalies occur in the propylitically altered areas around the **Snowfield Gold Zone** and on the West side of the **Ortum Fault**, indicating that the intensely sericitized **Snowfield Gossan** is depleted in zinc.

Isolated zinc spot anomalies occur around the **Coffee Pot Zone** which contains sphalerite in quartz veins in silicified sericite schists.

The **Coffee Pot Zone** is surrounded by a 150 x 100 m area of >3 ppm Ag and a broader area of >1 ppm. The east half of the grid has uniform 1-3 ppm Ag content similar in distribution to Mo values.

Cu anomalies are similar to Zn around the *Coffee Pot Zone*, but are absent around the *Snowfield Gold Zone*.

Poorly defined anomalous zones of >100 ppm As occurs around both the *Snowfield* and *Coffee Pot Zones*.

MINERALIZATION

SNOWFIELD GOLD ZONE

The *Snowfield Gold Zone* is located near the east end of the *Snowfield Gossan* in an area of foliated chloritized andesite tuffs. Sericitization is local and minor and pyrite content is 3-5%. The 1985 *Newhawk* drilling discovered the zone to have sharp assay boundaries, with Au values dropping abruptly from consistently .07 - .09 opt Au to 0.03 opt Au, but with no macroscopic change in lithology, alteration, or sulphide content. All five drill holes went through the gold zone into lower grade material. In situ resource was estimated by *Newhawk* as 7.7 million tons of 0.08 opt Au. Unlike most of the other Au mineralized zones in the *Sulphurets* area the *Snowfield Gold Zone* contains only trace Ag and base metals.

Mapping and airphoto interpretation indicated that the NE corner of the *Snowfield Gold Zone* is bounded by a northwest trending fault that dips 30° to the northeast, but the other boundaries of the zone are defined only by decreasing Au values.

A newly exposed nunatak 100 m south of the edge of the ice was sampled and found to carry >1000 ppb Au, indicating the possibility of greatly increasing reserves with further drilling in that direction.

COFFEE POT ZONE

A zone of silicified sericite schist was located during the lithogeochem sampling in the northernmost part of the grid. This was followed up by prospecting and further sampling. The Coffee Pot Zone consists of moderately-intense, pervasively silicified sericite pyrite schist cut by randomly oriented quartz veins. Massive pyrite with local tetrahedrite and chalcopyrite occurs locally in veins and in the silicified schist. The highest precious metal values obtained were .07 opt Au and 3.4 opt Ag from a high grade grab.

The zone appears to be in the footwall of a northwest trending northeast dipping fault and is cut off to the northeast by a major north-south shear. The zone is open to the southwest and further prospecting could extend the zone in that direction.

QUARTZ STOCKWORK ZONE

A 500 m x 500 m zone containing 30-50% quartz veins extends off the northeast corner of the grid. It contains only minor pyrite but molybdenite was discovered in the northeast part of the zone above the Mitchell Glacier by Granduc , and in 1968 two 1000' holes (DDH-7 and 8) were drilled to test the zone. No significant molybdenum mineralization was encountered. As the core had not originally been assayed for Au, it was resampled in 1989 and submitted for gold geochemical analysis. Au results are similar to the rest of the Snowfield Gossan in that almost all samples returned >200 ppb Au. Local zones of higher grade material were encountered, including a single section near the top of DDH-8 which returned .102 opt Au from a narrow oxidized shear. Background levels of Cu and Mo were high throughout the core.

The southernmost of the two holes, DDH-8, was collared in stockwork but quickly drilled out of it to the south. It appears that the zone is faulted off on the south and west sides.

MOLYBDENITE ZONE

A zone of significant molybdenite mineralization was discovered by Esso in early 1980s 300 m northwest of the Granduc drill holes. This area was probably snowcovered in 1968 but was intensely trenched by Esso. The zone extends from 1 N to 4 S between 2+50W and 4W and is comprised of sericite-pyrite schist with abundant molybdenite bearing quartz veins which returned Mo values up to .9%.

BARITE VEINS

Barite veins, hosted in tension gashes adjacent to fine grained mafic dykes occur in an indistinct zone which extends from 2+50W/4S to 5W/3S. The veins contain local galena and sphalerite and the highest value obtained was .04 opt Au and 0.32 opt Ag from a high grade grab.

One km southeast on the ridgetop is the Josephine Zone which contains spotty but spectacular Au and Ag values; up to 1.6 opt Au and 215 opt Ag, in quartz and barite veins. The two zones may be related but the intervening area is ice covered.

ORTUM FAULT

The best Au result during the 1989 work was .21 opt Au obtained from a 2-3 m wide milled quartz pyrite zone adjacent to the Ortum Fault, which terminates the Snowfield Gossan on the west side. Only minor prospecting was done during the 1989 program and any future work on the Snowfield Gossan area should include detailed prospecting and sampling along the fault.

CONCLUSIONS

The Mitchell-Sulphurets ridge is host to extensive alteration, considerable structural deformation and significant Au and base metal mineralization. Work to date has been sporadic and cursory, and although further work is unlikely to discover large obvious areas of surface mineralization,

potential does exist for uncovering mineralization of a more subtle nature.

Most of the *Snowfield Gossan* contains high Au values (>100 ppb), and though no additional *Snowfield Gold Zone* targets were located, it was shown that the zone does continue to the south under the ice. Mapping around the *Snowfield Gold Zone* and the *Snowfield Gossan* were only of a cursory nature and did not reveal any structural, alteration or lithologic controls to the Au mineralization.

The *Snowfield Gold Zone* differs markedly from other mineralized zone on the property, which are mesothermal, much less consistent in Au content and contain significant Ag.

Most of the *Mitchell-Sulphurets* ridge west of the grid is propylitically altered and although *Esso* did lithogeochemical sampling in 1980, some potential exists for the discovery of further mineralization by systematic grid sampling.

The mineralization on the *Coffee Pot Zone* is later than the above. It is hosted in faulted silicified sericite schists and contains base metal sulphides. The zone looked impressive but precious metal values were disappointing. It appears to be open to the southeast and further prospecting and sampling could extend it considerably.

The barite veins discovered in the south central part of the grid appear to postdate the fine grained mafic dykes and are probably the latest mineralizing event on the *Snowfield Gossan* area. Auriferous barite veins occur on the *Josephine Showing* 1 km to the southeast.

RECOMMENDATIONS

The existing grid should be extended to the north across the *Quartz Stockwork Zone* and to the *Moly Zone* on the *Mitchell Glacier*, and to the east as the *Snowfield Gold Zone* is open in that direction. To the west across the *Ortum*

Fault is a large area of propylitic altered volcanics and the potential exists for a discovering further Snowfield Gold Zone type mineralization.

Extending the grid and continuing the systematic lithogeochemical sampling will provide the basis for work on metal zonations to locate porphyry mineralization. Detailed mapping is required to understand controls on the known mineralization.

The retreat of snow and ice from the southern part of the Snowfield Gold Zone permits an opportunity for further drilling to the south which could significantly increase reserves there.

An obvious target for further sampling and detailed prospecting is the 7.3 g/t Au sample obtained from silicified and pyritic fault gouge on the Ortum Fault. This area has the best potential for an easily defined Au zone.

APPENDIX I

ROCK SAMPLE ANALYSES

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
3+00W 0+50N	77	338	40	20	1.0	17	8	9	2.90	7	5	ND	1	12	1	2	2	2	.03	.008	2	1	.01	8	.01	3	.24	.01	.10	2	134
3+00W 0BL	192	559	74	301	1.2	57	22	276	4.11	14	5	ND	3	128	1	2	2	22	.08	.096	2	20	.56	23	.01	2	1.40	.01	.10	1	1040
3+00W 0+50S	97	291	107	481	1.2	35	10	359	4.09	6	5	ND	1	21	5	2	2	18	.10	.076	2	12	.56	9	.01	6	1.43	.01	.07	1	400
3+00W 1+00S	81	139	51	76	1.1	22	17	61	4.90	20	5	ND	2	8	1	2	2	5	.15	.068	2	2	.10	4	.01	2	.43	.01	.13	1	430
3+00W 1+50S	74	429	75	145	1.5	13	9	362	4.09	165	5	ND	1	8	2	4	2	7	.28	.064	2	3	.26	6	.01	2	.67	.01	.11	1	430
3+00W 3+50S	44	218	43	47	.6	30	12	90	3.76	19	5	ND	1	14	2	2	3	7	.16	.062	4	13	.28	6	.01	14	.54	.01	.17	2	101
2+90W 2+00N	237	144	27	124	.8	20	10	114	4.38	11	5	ND	1	4	1	2	2	10	.04	.030	2	5	.21	8	.01	2	.67	.01	.06	1	420
2+60W 3+50S	26	719	30	107	.4	22	9	332	3.54	2	5	ND	3	15	1	2	2	29	.23	.093	6	19	2.21	10	.01	3	2.09	.01	.14	1	144
2+00W 1+50N	225	316	97	109	1.0	5	2	53	8.00	18	5	ND	1	14	1	2	2	6	.01	.056	2	3	.01	101	.01	5	.31	.01	.10	1	320
2+00W 1+00N	177	105	171	97	1.6	5	2	61	2.47	16	5	ND	1	18	1	2	2	4	.01	.036	2	2	.10	113	.01	2	.42	.01	.11	1	320
2+00W 0+50N	228	168	92	370	1.4	6	5	1896	3.81	24	5	ND	2	36	2	2	2	27	.06	.064	2	10	.66	30	.01	12	1.11	.01	.10	1	1460
2+00W 0BL	202	233	53	1099	1.5	4	13	2406	7.45	32	5	3	3	10	2	2	2	93	.39	.164	6	11	2.49	15	.01	3	3.59	.01	.09	1	1960
2+00W 0+50S	35	222	23	1778	.4	3	6	1502	4.28	13	5	ND	2	29	12	2	2	25	1.64	.082	2	8	1.43	26	.01	2	2.27	.01	.10	1	1380
2+00W 1+00S	42	281	45	662	.7	3	10	2661	4.40	17	5	ND	2	52	2	2	2	26	2.85	.126	4	8	1.00	15	.01	8	1.73	.01	.11	1	930
0+00 9+15W	3	26	550	34	54.9	13	87	309	15.72	615	7	ND	1	15	1	2	3	3	.22	.005	2	9	.03	1	.01	10	.15	.01	.10	8	7340
0+00 9+00W	1	10	38	57	1.8	12	7	702	2.68	46	5	ND	1	16	1	2	2	12	.32	.061	8	12	.46	16	.03	2	.57	.02	.13	3	156
0+00 8+50W	1	17	98	159	1.2	4	11	1424	3.93	29	6	ND	3	11	1	2	2	27	.28	.114	7	9	.94	9	.07	2	1.15	.01	.23	6	91
0+00 8+00W	1	45	165	289	2.5	4	9	1077	5.19	69	5	ND	1	50	1	2	2	28	.20	.125	5	10	1.33	7	.01	2	1.46	.01	.19	1	106
0+00 7+50W	10	349	54	147	1.2	18	14	375	3.94	41	5	ND	1	14	1	2	2	16	.06	.046	4	15	1.47	6	.01	5	1.37	.01	.15	1	129
0+00 7+00W	8	163	243	246	2.5	13	6	200	3.76	32	5	ND	1	12	3	3	2	11	.06	.055	2	10	.78	12	.01	12	.84	.01	.14	1	66
0+00 6+50W	2	48	89	9	.8	23	15	15	4.94	15	5	ND	1	6	1	2	2	4	.09	.049	2	1	.02	5	.01	12	.27	.01	.12	1	96
0+00 5+00W	150	188	37	87	.5	15	9	251	3.80	2	5	ND	1	11	1	2	2	13	.09	.054	3	14	1.03	12	.01	2	1.12	.01	.12	1	200
0+00 2+50W	191	96	74	98	.8	9	4	111	1.89	54	5	ND	1	24	1	4	2	4	.01	.009	2	4	.08	15	.01	3	.32	.01	.09	1	141
1+05S 8+08W	5	508	425	57	10.6	7	5	23	16.34	72	7	ND	1	5	1	54	12	2	.04	.021	2	5	.01	1	.01	3	.16	.01	.09	1	320
1+40S 5+20W	6	33	40	116	.5	9	1	314	1.44	2	5	ND	1	74	1	2	3	7	.01	.005	2	14	1.23	25	.01	5	.98	.01	.02	1	43
3+10S 4+70W	10	24	41	16	.5	18	4	29	2.32	6	5	ND	1	20	1	2	2	3	.03	.017	2	4	.07	9	.01	2	.29	.01	.08	1	110
STD C/AU-R	18	58	43	132	6.6	67	30	946	4.11	39	22	7	37	48	19	15	22	58	.48	.094	38	54	.88	175	.06	34	1.94	.06	.14	12	480

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	Tl %	B PPM	Al %	Na %	K %	W PPM	Au* PPB
1+00E 2+00N	149	94	8	80	1.2	7	2	18	1.78	10	5	ND	1	25	1	7	2	4	.01	.005	2	8	.01	283	.01	4	.17	.01	.08	1	400
1+00E 1+50N	103	35	149	60	2.1	6	2	17	1.57	8	5	ND	1	39	1	2	4	6	.01	.041	2	7	.01	702	.01	2	.30	.01	.15	1	390
1+00E 0+50S	209	86	33	594	1.2	4	3	1269	3.87	83	5	2	4	7	3	17	2	50	.12	.093	6	12	1.44	234	.01	3	1.72	.01	.13	1	2070
1+00E 1+00S	42	249	36	769	2.2	7	15	2618	5.08	97	8	3	7	12	4	49	10	57	.43	.115	6	12	2.01	52	.01	4	2.62	.01	.13	1	2100
1+00E-1+50S	103	67	53	634	1.2	4	6	1906	4.30	205	5	3	3	12	5	21	8	39	.33	.121	9	6	1.00	29	.01	11	1.36	.01	.12	1	2200
1+00E 1+55S	84	58	29	817	.6	4	7	4506	3.76	34	5	ND	3	37	7	11	2	58	2.34	.111	7	8	1.70	71	.01	2	2.27	.01	.14	1	670
STD C/AU-R	18	61	42	131	7.0	68	31	1022	3.97	4.1	22	8	40	49	19	16	21	60	.50	.091	40	55	.89	174	.06	35	2.04	.06	.14	13	510

Assay Recommended for Cu, Pb, Zn > 1%
Ag > 30 ppm

GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER.
 THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: ROCK AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

DATE RECEIVED: OCT 3 1989 DATE REPORT MAILED: *Oct 11/89* SIGNED BY: *C. Leong* D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

Corona Corporation PROJECT 1014 File # 89-4062

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
8+20W 2+00N	12	32	63	51	2.8	21	9	187	8.73	140	5	ND	2	17	1	2	6	8	.38	.017	2	16	.03	2	.01	4	.23	.02	.16	3	370
8+15W 1+45N	138	220	97	243	2.4	9	24	259	7.64	123	6	ND	1	30	2	2	2	4	.43	.018	2	4	.02	1	.01	2	.20	.01	.14	1	310
8+00W 1+75N	4	130	416	706	2.1	36	32	1025	6.15	107	5	ND	3	37	5	8	2	27	.96	.077	4	20	.59	7	.01	2	.69	.02	.24	3	360
7+85W 1+50N	11	3773	617	2521	36.5	10	16	585	11.31	925	5	2	3	75	28	138	5	7	.64	.031	3	3	.04	3	.01	3	.19	.01	.14	1	1710
7+75W 2+05N	2	41	172	390	1.9	25	13	361	4.11	92	5	ND	2	28	3	2	2	8	.45	.044	2	13	.11	8	.01	10	.29	.01	.18	2	185
7+70W 1+35N	2	178	93	20	14.5	15	11	16	6.58	58	5	ND	2	14	1	13	2	14	.04	.018	4	5	.01	4	.01	2	.21	.01	.20	1	470
7+50W 1+25N	3	602	78	45	3.0	6	4	416	1.73	17	5	ND	1	31	1	2	2	7	.29	.024	3	19	.04	57	.01	4	.27	.01	.22	1	97
7+35W 1+75N	2	166	1552	1541	118.7	4	12	18	9.45	150	5	3	3	20	13	14	22	8	.02	.019	3	5	.01	2	.01	4	.18	.01	.18	1	2110
7+35W 1+78N	1	2117	2886	1658	62.4	5	8	196	7.84	918	5	3	3	58	19	66	18	10	.37	.101	4	14	.03	6	.01	3	.23	.01	.18	1	2430
7+35W 2+40N	2	877	1357	5796	16.5	3	7	18	7.15	387	5	ND	3	22	53	117	2	9	.01	.011	6	4	.01	2	.01	2	.19	.01	.15	3	133
7+25W 2+25N	1	139	64	87	1.9	2	6	4269	2.87	71	5	ND	1	263	1	15	5	6	2.57	.075	3	40	.05	18	.01	16	.20	.01	.14	1	58
7+10W 1+40N	2	555	301	70	3.8	11	3	1012	1.57	8	5	ND	2	131	1	2	2	8	.57	.054	4	6	.24	59	.01	9	.30	.01	.21	1	132
7+05W 1+25N	3	617	372	92	5.8	20	12	731	4.55	50	5	ND	1	81	1	2	2	18	.50	.071	2	21	.16	9	.01	6	.35	.01	.24	1	400
7+05W 1+55N	1	762	375	426	6.9	3	4	9185	3.08	170	5	ND	2	534	5	40	2	10	4.78	.048	3	5	1.80	12	.01	6	.21	.01	.14	1	124
7+00W 2+10N	3	168	82	20	7.2	2	13	100	10.57	152	5	ND	3	15	1	2	9	9	.14	.062	3	14	.03	3	.01	2	.23	.01	.18	1	570
6+30W 1+15N	5	516	60	50	1.8	3	14	294	3.68	46	5	ND	2	73	1	16	2	17	.36	.117	3	3	.38	4	.01	6	.46	.01	.22	1	74
6+30W 1+16N	4	19	34	8	.3	7	1	74	1.22	29	5	ND	1	167	1	2	2	3	.01	.020	2	8	.01	103	.01	2	.07	.01	.04	1	19
6+30W 1+25N	3	244	22	27	.9	12	5	32	1.86	26	5	ND	1	43	1	2	2	7	.08	.048	3	10	.03	16	.01	3	.22	.02	.16	1	97
6+45W 1+30N	5	1080	51	89	7.6	12	9	141	3.98	160	5	ND	1	31	1	93	2	9	.15	.045	2	5	.10	3	.01	7	.27	.01	.18	1	660
6+10W 0+80N	1	1204	98	206	7.2	22	107	89	14.47	278	5	ND	2	4	2	11	2	24	.11	.045	2	9	1.64	1	.01	4	2.02	.01	.06	1	380
6+00W 1+75N	1	248	29	104	1.0	2	5	157	4.22	43	5	ND	4	34	1	2	2	21	.15	.130	5	2	.29	29	.01	6	.63	.01	.23	1	81
5+25W 0+50N	787	129	127	153	3.3	8	23	21	10.27	82	5	ND	2	55	2	4	2	6	.02	.015	2	4	.08	3	.01	2	.25	.01	.07	1	410
5+70W 3+05S	1	16	63	5	.4	8	5	12	1.88	9	5	ND	1	25	1	2	2	2	.03	.025	2	2	.01	14	.01	4	.25	.01	.14	1	27
5+55W 2+75S	5	4	19	2	.1	2	1	8	.59	6	5	ND	1	13	1	2	2	3	.01	.016	2	5	.01	264	.01	4	.26	.01	.16	1	92
5+12W 2+70S	1	1016	13	256	1.9	27	22	782	10.72	12	5	ND	3	5	1	2	2	57	.46	.187	3	14	2.33	7	.01	6	2.70	.01	.10	1	390
5+00W 3+20S	5	4	13	8	.1	7	1	2	.76	3	5	ND	1	318	1	2	2	1	.01	.030	2	8	.01	36	.01	3	.05	.01	.03	1	25
3+50W 2+45S	44	84	45	24	.4	9	6	13	2.65	13	5	ND	1	17	1	19	2	2	.05	.023	2	4	.04	8	.01	5	.27	.01	.10	1	116
3+50W 2+25S	119	60	40	30	.8	9	5	7	2.24	9	5	ND	1	24	1	18	2	4	.01	.012	2	4	.01	8	.01	5	.23	.01	.12	1	870
3+50W 2+18S	1	68	9	209	.1	66	25	698	4.10	34	5	ND	1	252	1	2	2	84	2.78	.123	6	152	2.74	276	.18	8	2.66	.11	.09	1	12
3+50W 2+00S	134	133	146	126	.6	21	14	59	4.51	40	5	ND	1	7	1	22	2	9	.13	.048	2	8	.11	8	.01	5	.48	.01	.13	1	420
3+50W 1+75S	117	265	219	154	.5	12	11	22	3.67	34	5	ND	1	45	2	2	2	5	.03	.022	2	4	.10	6	.01	3	.36	.01	.10	1	360
3+50W 1+50S	88	166	44	47	.8	17	8	11	3.46	37	5	ND	1	12	1	2	2	4	.02	.030	2	4	.06	9	.01	3	.39	.01	.14	1	700
3+50W 1+25S	216	152	67	59	1.1	59	17	27	6.52	114	5	2	2	5	1	2	2	5	.10	.044	2	3	.05	6	.01	3	.34	.01	.12	1	1030
STD C/AU-R	18	63	36	132	6.7	68	31	1008	4.00	41	24	7	38	48	18	16	22	59	.49	.089	38	56	.89	174	.06	34	1.92	.06	.13	13	520

✓ ASSAY RECOMMENDED

ACME ANALYTICAL LABORATORIES LTD.
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE(604)253-3158 FAX(604)253-1716

DATE RECEIVED: OCT 10 1989

DATE REPORT MAILED: *Oct 16/89*

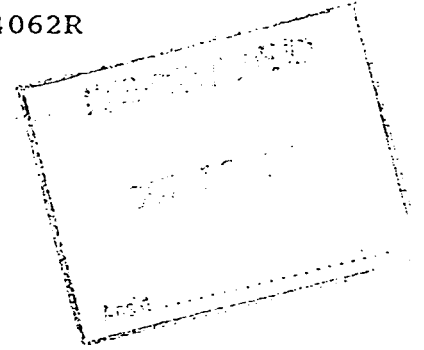
ASSAY CERTIFICATE

AU** BY FIRE ASSAY FROM 1/2 A.T.
- SAMPLE TYPE: ROCK PULP

SIGNED BY *C. King* D.TOYE, C.LEONG, J.WANG; CERTIFIED B.C. ASSAYERS

Corona Corporation PROJECT 1014 FILE # 89-4062R

SAMPLE#	Au**
	OZ/T
7+85W 1+50N	.053
7+35W 1+75N	.075
7+35W 1+78N	.078
3+50W 1+25S	.032



APPENDIX II

DRILL LOGS - DDH 7, 8

DDH-7

Drilled 1968 Granduc Mines Ltd. Depth: 306.32m Collar Elevation: 1590m
 Grid: 0+20E/4+00N AZ: 180° Dip: -45°(?)

<u>Interval</u>	<u>Description</u>	<u>Interval</u>	<u>Sample</u>	<u>Au(ppb)</u>	<u>Mo(ppm)</u>	
6.10-306.32	qtz stockwork, 20-50% 1-2 cm white qtz veins with minor py, trace mo, cp, in qtz-ser schist	6.10- 9.02m	20501	300	46	
		- 12.19	02	320	51	
		- 15.54	03	280	81	
		- 18.25	04	300	66	
		- 21.34	05	300	72	
6.10-25.0		20-50% qtz veins	- 24.08	06	200	49
			- 28.04	07	410	58
7.62		ox shear	- 30.18	08	230	71
			- 32.00	09	430	103
23.7-24.08		fault gouge	- 35.05	10	370	103
			- 38.10	11	200	84
25.0-306.32		20-30% qzt veins	- 39.47	12	190	77
			- 42.67	13	320	101
141.73-142.23		fault gouge	- 45.87	14	310	71
			- 49.09	15	390	97
157.58-158.19		fault gouge	- 52.27	16	290	121
			- 55.47	17	370	95
171.0			- 58.52	18	430	66
		99.06-101.19	19	200	122	
		95.40- 99.06	20	390	92	
306.32	E.O.H.	92.35- 95.40	21	530	95	
		89.64- 92.35	22	118	20	
		87.19- 89.64	23	640	82	
		84.12- 87.19	24	650	61	
		80.92- 84.12	25	620	55	
		77.60- 80.92	26	290	79	
		74.37- 77.60	27	830	73	
		71.17- 74.37	28	780	76	
		67.97- 71.17	29	640	89	
		64.92- 67.97	30	390	96	
		61.57- 64.92	31	450	77	
		58.52- 61.57	32	500	72	
		101.19-109.73	33	430	82	
		170.99-174.04	34	530	120	
		167.94-170.99	35	1020	101	
		164.59-167.94	36	340	113	
		161.57-164.59	37	470	107	
		158.50-161.57	38	270	107	
		155.45-158.50	39	520	108	
		152.86-155.45	40	710	75	
		148.71-152.86	41	370	71	
		145.54-148.71	42	300	65	
		141.73-145.54	43	123	68	
		138.68-141.73	44	240	85	

DDH-7 (Continued)

<u>Interval</u>	<u>Description</u>	<u>Interval</u>	<u>Sample</u>	<u>Au(ppb)</u>	<u>Mo(ppm)</u>
		135.64-138.68m	20545	200	77
		131.98-135.64	46	128	90
		128.93-131.98	47	178	78
		125.88-128.93	48	124	76
		122.22-125.88	49	149	95
		117.65-122.22	50	135	100
		239.57-242.64	51	490	76
		236.52-239.57	52	380	93
		233.48-236.52	53	600	97
		230.43-233.48	54	450	85
		227.38-230.43	55	380	98
		224.33-227.38	56	178	90
		221.28-224.33	57	177	98
		218.24-221.28	58	230	81
		215.18-218.24	59	540	118
		212.14-215.18	60	380	157
		209.09-212.14	61	164	68
		206.04-209.09	62	290	70
		109.73-113.39	63	320	87
		113.39-117.65	64	350	98
		203.00-206.04	65	430	104
		199.95-203.00	66	520	98
		196.90-199.95	67	750	94
		193.85-196.90	68	920	97
		190.80-193.85	69	650	160
		187.76-190.80	70	610	97
		184.71-187.76	71	650	106
		181.66-184.71	72	800	119
		178.61-181.66	73	820	171
		174.04-178.61	74	650	108
		303.28-306.32	75	220	61
		300.23-303.28	76	260	67
		297.48-300.23	77	151	65
		294.44-297.48	78	153	72
		291.39-294.44	79	380	66
		288.34-291.39	80	99	60
		285.29-288.34	81	120	49
		282.24-285.29	82	124	45
		279.20-282.24	83	128	68
		276.15-279.20	84	138	62
		272.80-276.15	85	190	58
		270.05-272.80	86	146	74
		267.00-270.05	87	145	75
		263.96-267.00	88	164	96
		260.91-263.96	89	185	90
		257.86-260.91	90	200	108
		254.81-257.86	91	240	71
		251.76-254.81	92	120	74

DDH-7 (Continued)

<u>Interval</u>	<u>Description</u>	<u>Interval</u>	<u>Sample</u>	<u>Au(ppb)</u>	<u>Mo(ppm)</u>
		248.72-251.76m	20593	380	83
		245.67-248.72	94	850	64
		242.62-245.67	95	760	61

DDH-8

Drilled 1968 Granduc Mines Ltd. Depth: 304.80m Collar Elevation: 1650m
 Grid: 0+00W/1+95N AZ: 180° Dip: -45°(?)

<u>Interval</u>	<u>Description</u>	<u>Interval</u>	<u>Sample</u>	<u>Au(ppb)</u>	<u>Mo(ppm)</u>
1.21-182.0	light gy ser sch, local massive sections. 1-3% diss, stringer py, local minor mo on fractures & qtz veins. Foliation 70-90° CA	1.21- 6.06m	20401	1240	72
		- 7.92	02	1300	53
		- 10.97	03	990	43
		- 14.02	04	3490	101
		- 15.54	05	660	78
		- 18.59	06	870	116
		- 22.55	07	450	149
14.6	ox shear	- 26.21	08	610	69
		- 29.26	09	150	75
13.59- 26.21	50% recovery, ox shear	- 32.31	10	200	109
		- 35.36	11	450	77
30.78	ox shear	- 38.40	12	500	99
		- 41.18	13	750	121
27.0-57.0	qtz stockwork; 10-15 veins/m	- 44.50	14	350	113
		- 47.24	15	380	82
		- 49.37	16	280	80
52.12-62.18	ox shear	- 52.12	17	320	103
		- 57.00	18	300	126
171.0	slickensides 45° CA	- 62.18	19	380	126
		- 65.23	20	370	60
		- 67.36	21	690	98
182.0-304.80	chloritized fragmental andesite - lapilli tuff?; up to 1% diss py, local chl-py veins, 70-80° CA foliation	- 70.41	22	370	50
		- 73.76	23	510	73
		- 76.81	24	480	71
		- 79.86	25	420	74
		- 82.91	26	380	67
		- 85.95	27	430	91
		- 89.00	28	780	85
304.80	E.O.H.	- 92.04	29	620	82
		- 95.10	30	510	69
		- 98.14	31	590	108
		-101.19	32	390	104
		-103.63	33	270	115
		-106.83	34	520	92
		-109.42	35	360	90
		-112.57	36	500	80
		112.57-115.52	37	350	60
		-118.57	38	440	95
		-121.77	39	380	110
		-124.97	40	400	165
		-128.17	41	320	80
		-129.23	42	330	86
		-132.44	43	124	65
		-135.64	44	181	98

DDH-8 (Continued)

<u>Interval</u>	<u>Description</u>	<u>Interval</u>	<u>Sample</u>	<u>Au(ppb)</u>	<u>Mo(ppm)</u>
		135.64-138.84m	20445	210	76
		-140.21	46	157	114
		-143.25	47	132	126
		-146.30	48	270	70
		-149.35	49	290	94
		-152.40	50	310	130
		-155.45	51	480	73
		-158.50	52	163	82
		-161.54	53	153	73
		-164.90	54	430	84
		-167.94	55	380	85
		-171.14	56	410	85
		-174.19	57	340	127
		-177.39	58	340	83
		-180.59	59	230	90
		-183.79	60	470	89
		-186.84	61	700	81
		-189.89	62	720	86
		-192.24	63	710	77
		-197.05	64	710	89
		-200.10	65	270	148
		-203.15	66	220	88
		-206.20	67	720	94
		-209.09	68	1070	92
		-212.14	69	1120	92
		-215.19	70	470	98
		-218.24	71	520	98
		-221.28	72	310	94
		-224.33	73	227	95
		-227.08	74	440	78
		-230.43	75	330	73
		-233.48	76	700	92
		-238.35	77	980	103
		-241.40	78	900	135
		-244.45	79	850	109
		-247.50	80	990	95
		-250.54	81	760	102
		-253.59	82	630	98
		-256.79	83	580	75
		-260.91	84	880	69
		-263.96	85	390	68
		-266.70	86	540	59
		-270.05	87	450	80
		-273.40	88	570	56
		-276.15	89	320	129
		-279.20	90	91	70
		-282.24	91	970	103
		-285.29	92	1090	82

DDH-8 (Continued)

<u>Interval</u>	<u>Description</u>	<u>Interval</u>	<u>Sample</u>	<u>Au(ppb)</u>	<u>Mo(ppm)</u>
		135.64-288.34m	20493	880	129
		-291.38	94	560	127
		-294.43	95	940	129
		-297.48	96	770	136
		-300.53	97	670	134
		-303.58	98	700	75
		-304.80	99	390	67

APPENDIX III
DRILL CORE ANALYSES

GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER. THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM. - SAMPLE TYPE: Core AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

DATE RECEIVED: OCT 3 1989 DATE REPORT MAILED: Oct 11/89 SIGNED BY: D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

Corona Corporation PROJECT 1014 File # 89-4061 Page 1

Table with columns for SAMPLE#, elements (Mo, Cu, Pb, Zn, Ag, Ni, Co, Mn, Fe, As, U, Au, Th, Sr, Cd, Sb, Bi, V, Ca, P, La, Cr, Mg, Ba, Ti, B, Al, Na, K, W), and Au* (PPM). It lists data for samples E 20401 through E 20436, plus a standard sample STD C/AU-R.

Corona Corporation PROJECT 1014 FILE # 89-4061

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Ni PPH	Co PPM	Mn PPM	Fe %	As PPH	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
E 20682	45	875	57	277	1.1	13	8	198	3.97	2	5	ND	1	6	2	2	2	4	.17	.060	2	8	.27	22	.01	3	.54	.01	.09	1	124
E 20683	68	1144	73	294	1.2	16	8	61	4.00	9	5	ND	1	11	2	2	3	.25	.087	2	6	.09	17	.01	4	.30	.01	.10	1	128	
E 20684	62	1035	48	277	1.1	15	8	61	3.92	12	5	ND	1	9	2	2	2	.09	.029	2	5	.06	17	.01	5	.25	.01	.10	1	138	
E 20685	58	908	49	486	1.5	15	7	47	3.50	35	5	ND	1	8	2	9	3	2	.06	.017	2	3	.06	18	.01	3	.24	.01	.10	1	190
E 20686	74	1163	62	659	1.3	16	7	49	3.83	14	5	ND	1	7	3	2	3	2	.11	.031	2	6	.04	18	.01	7	.25	.01	.11	1	146
E 20687	75	1021	53	342	1.4	13	6	42	3.16	14	5	ND	1	9	2	3	2	2	.07	.011	2	6	.04	21	.01	4	.22	.01	.10	1	145
E 20688	96	930	51	396	.9	16	7	44	3.58	24	5	ND	1	9	2	8	3	2	.12	.038	2	6	.05	16	.01	5	.25	.01	.10	1	164
E 20689	90	845	66	259	1.2	15	7	73	2.96	7	5	ND	1	7	2	2	2	2	.08	.018	2	4	.06	21	.01	3	.22	.01	.09	1	185
E 20690	108	891	49	454	1.0	18	9	48	3.27	4	5	ND	1	15	3	2	2	3	.09	.020	2	6	.06	18	.01	2	.23	.01	.09	1	200
E 20691	71	836	74	256	1.1	17	6	46	3.23	12	5	ND	1	15	2	2	2	3	.07	.018	2	5	.06	19	.01	4	.26	.01	.10	1	240
E 20692	74	700	105	219	1.4	17	7	38	2.98	10	5	ND	1	21	2	2	2	2	.10	.024	2	6	.04	17	.01	5	.23	.01	.10	1	120
E 20693	83	1216	67	387	2.2	24	10	55	4.32	2	5	ND	1	7	2	2	3	3	.13	.042	2	7	.06	15	.01	3	.28	.01	.11	1	380
E 20694	64	854	35	435	1.6	25	9	155	4.32	7	5	ND	1	6	3	2	2	5	.10	.031	2	8	.15	16	.01	5	.38	.01	.10	1	850
E 20695	61	763	8	412	1.4	34	12	741	5.53	4	5	ND	2	5	1	2	2	17	.20	.079	2	21	.86	27	.01	5	1.51	.01	.11	1	760
STD C/AU-R	17	58	42	134	6.6	67	30	1034	4.13	37	20	7	38	48	18	15	23	58	.48	.093	38	58	.88	174	.06	37	1.92	.06	.13	11	520

ACME ANALYTICAL LABORATORIES LTD.
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE(604)253-3158 FAX(604)253-1716

DATE RECEIVED: OCT 12 1989

Oct 16/89

DATE REPORT MAILED:

ASSAY CERTIFICATE

AU** BY FIRE ASSAY FROM 1/2 A.T.
- SAMPLE TYPE: ROCK PULP

SIGNED BY *C. Chung* D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

Corona Corporation PROJECT 1014 FILE # 89-4061R

SAMPLE#	Au** OZ/T
E 20401	.039
E 20402	.037
E 20404	.105
E 20468	.030
E 20469	.032
E 20492	.033
E 20535	.036

