

GEOLOGICAL SURVEY BRANCH
ASSESSMENT REPORT

28,278

PEROVIC ENTERPRISES INC.

The Tay-Christina Gold Property
Report on 2006 Exploration Drilling Program

Alberni Mining Division
NTS 92F/6W BCGS 92F. 034

Lat 49° 20' North
Long 125° 15' West

UTM Zone 10 U
Easting 0334600 Northing 5463100

Owner: Franjo Perovic
Operator: Perovic Enterprises Inc.

Author: Franjo Perovic
Consultant: Alf Randall, P.Eng

April 10, 2006

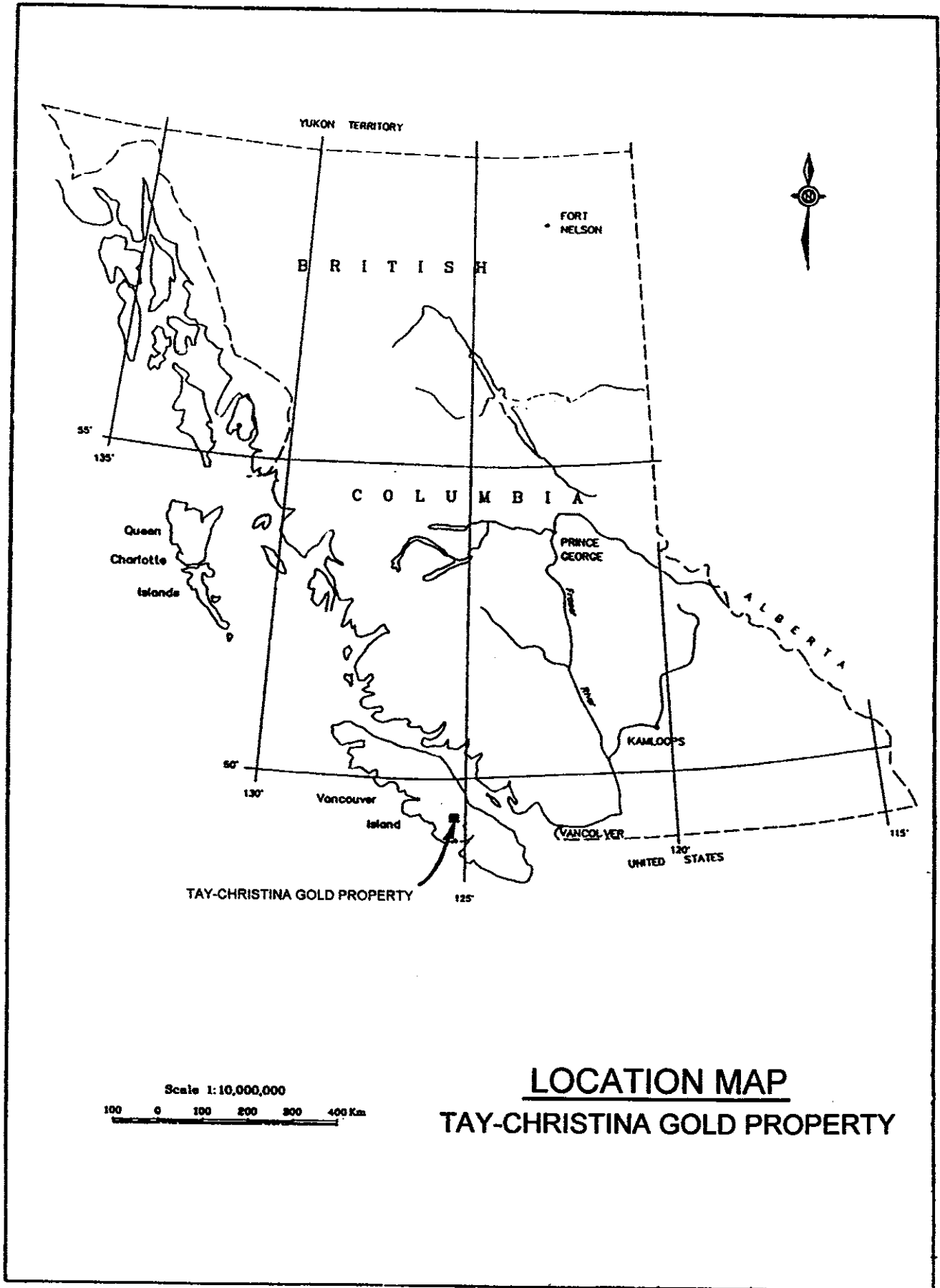


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LOCATION MAP
TAY-CHRISTINA GOLD PROPERTY

Figure 1 Location Map

VANCOUVER ISLAND

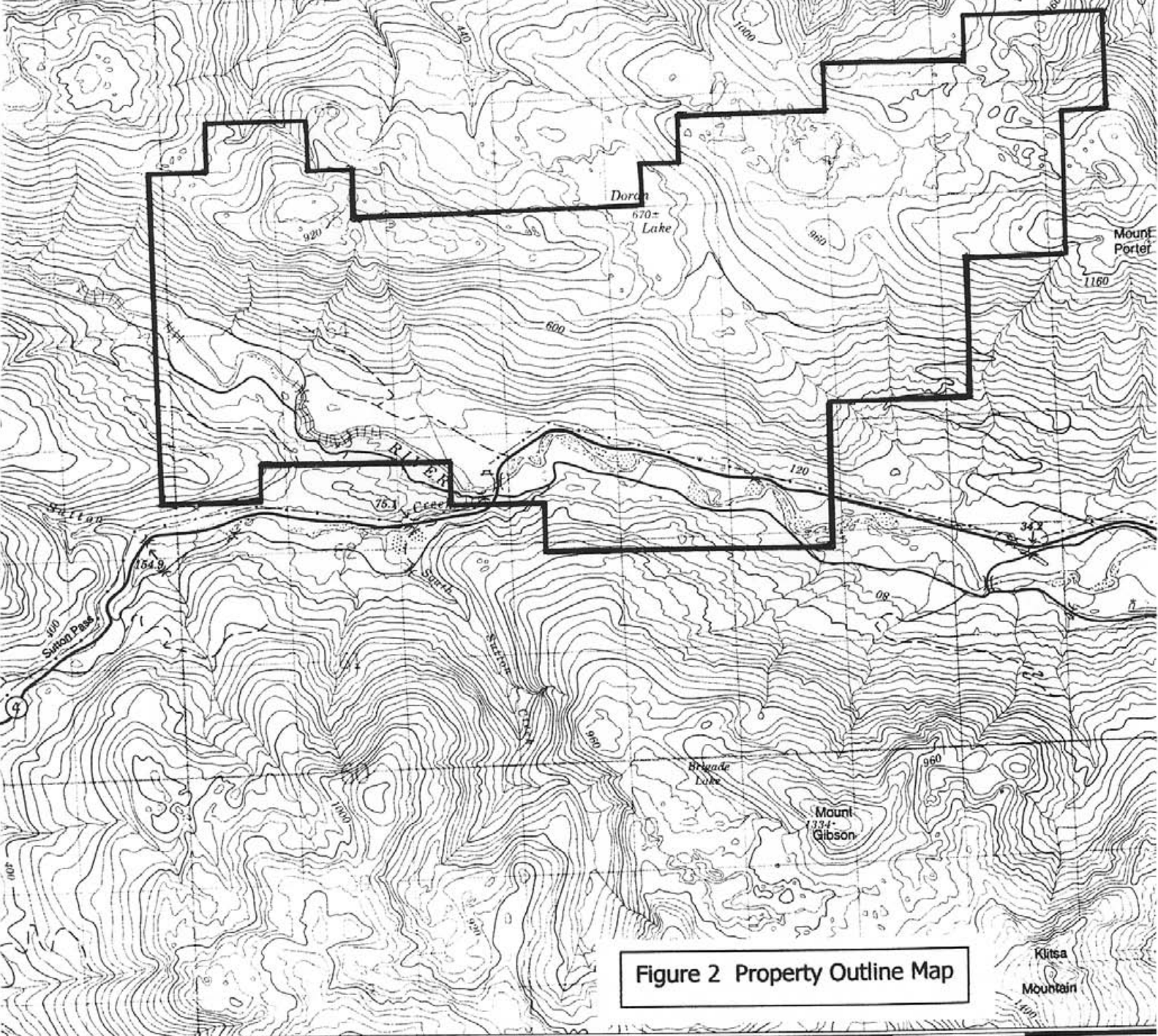


Figure 2 Property Outline Map

20' 14 15 15'

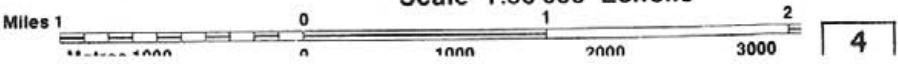
NTS 92F06



GREAT CENTRAL LAKE BRITISH COLUMBIA COLOMBIE-BRITANNIQUE

Routes:		
revêtement dur dual highway more than 2 lanes
revêtement dur double chaussée plus de 2 voies
..... 2 lanes less than 2 lanes
..... 2 voies moins de 2 voies
gravier, aggloméré, toute saison 2 lanes or more less than 2 lanes
..... 2 voies ou plus moins de 2 voies
de gravier, temps sec		
route non classée ou rue		
de terre		
sentier, percée ou portage		

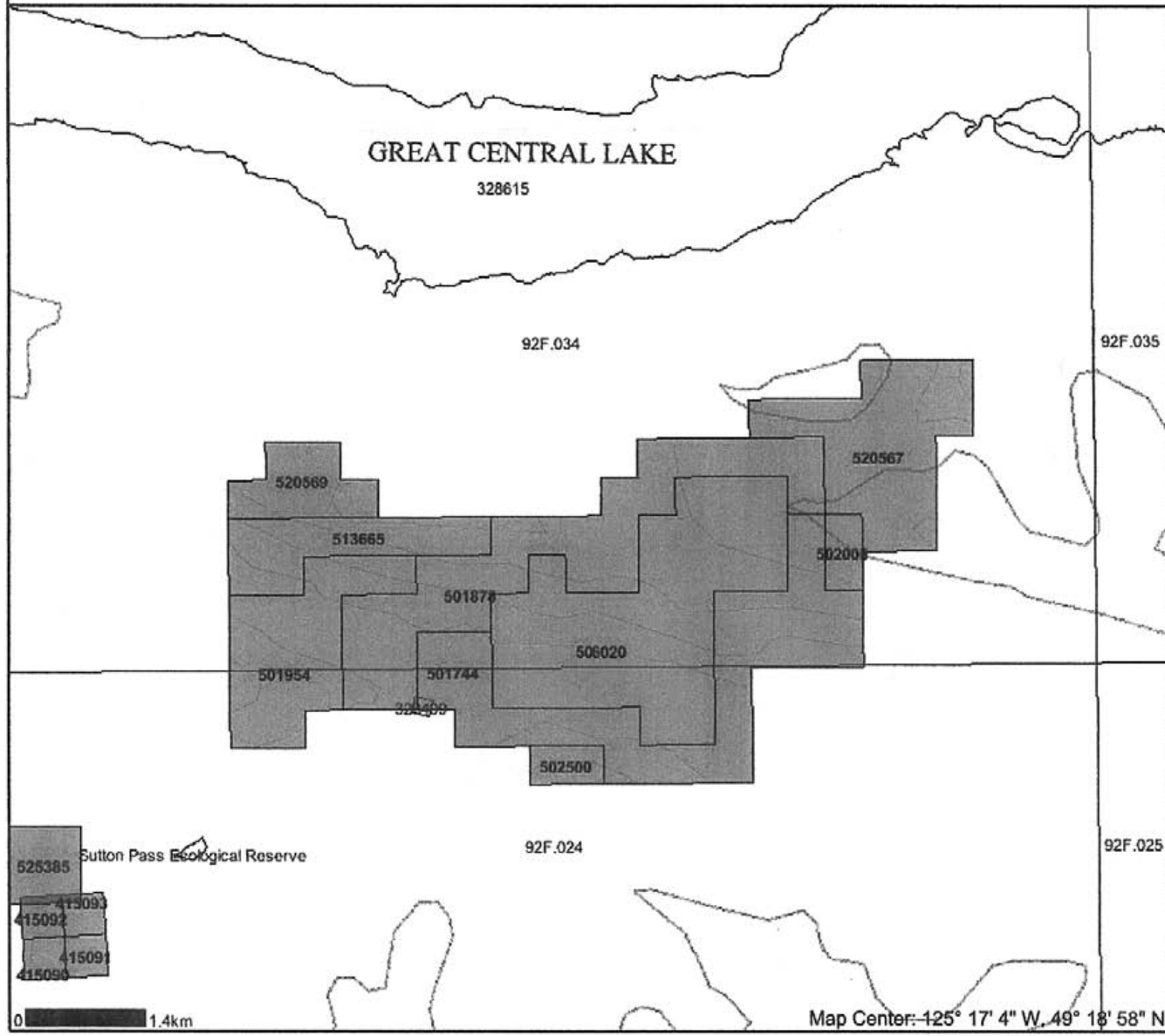
Scale 1:50 000 Échelle



Map created Sat Mar 25 15:39:59 PST 2006

Legend

- Indian Reserves
- National Parks
- Parks
- Mineral Tenures Reserves (Sites)
- Placer Claim Designation
- Placer Lease Designation
- No Staking Reserve
- Conditional Reserve
- Release Required Reserve
- Surface Restriction
- Recreation Area
- Others
- Mining Divisions
- BCGS Grid
- Contours (1:250K)
- Contour - Index
- Contour - Intermediate
- Area of Exclusion
- Area of Indefinite Contours
- Major Cities



Scale: 1:75,000

DO NOT USE FOR NAVIGATION

Figure 3 Mineral Tenure Map

INTRODUCTION

The Tay-Christina Gold Property, formerly known as the Tay Property, is currently owned by Franjo Perovic and operated by Perovic Enterprises Inc. The location is at the Taylor River, 40 kilometers west of Port Alberni along highway #4, in the Alberni Mining Division of Vancouver Island.

The property history and economic assessment is included in the consultants report. (See "Interpretation of results relative to geology and conclusion" pages 8-12).

OBJECTIVE AND SCOPE OF PRESENT WORK

Using a manportable prospector diamond drill with the potential to retrieve 36.4 mm diameter core samples, we endeavored to explore for a vein system which was associated with a batholith intrusion, a fault and a gold in soil anomaly. Two holes were drilled totaling 370.64 meters in length within the boundaries of the Mineral Tenure #506020. The core samples are located at Junction Self Storage Ltd., 2530 Timberlane Rd., Port Alberni, BC.

The work commenced on April 17, 2005 and carried out mostly through to May 22, 2005. In order to complete site reclamation, geological consulting, core examination, comparisons to previous sampling, prospecting and core logging, the work continued up to and including December 27, 2005. The program was carried out under our Notice of Work, Mine No.: 1610155.

Due to limited time and budget, the zone which was intended to be covered by exploration drilling was only partially completed. The source for the gold in soil anomaly was not identified.

It is recommended that the exploration continues and that the remaining portion of the target area be tested based on evidence of hydrothermal activity in the core samples retrieved, minor sulphides and various reports which elaborate on the above mentioned geological data. (See "References cited in this report")

The Property is comprised of the following claims:

(See following page)

Tenure Number	Tenure Type	Claim Name	Owner	Map Number	Good To Date	Status	Mining Division	Area	Tag Number
501744	Mineral	BFTGOG 1	143173 (100%)	092F	2011/JAN/12	GOOD		526.647	
501878	Mineral	BFTGOG 2	143173 (100%)	092F	2011/JAN/12	GOOD		526.518	
501954	Mineral	BFTGOG 3	143173 (100%)	092F	2010/JAN/12	GOOD		294.901	
502005	Mineral	BFTGOG 4	143173 (100%)	092F	2010/JAN/12	GOOD		42.118	
502500	Mineral	BFTGOG 5	143173 (100%)	092F	2010/JAN/12	GOOD		42.14	
506020	Mineral		143173 (100%)	092F	2013/DEC/16	GOOD		674.03	
513665	Mineral	BFTGOGTGI	143173 (100%)	092F	2006/MAY/31	GOOD		189.539	
520567	Mineral	BFTGOGBILQ	143173 (100%)	092F	2006/SEP/28	GOOD		357.917	
520569	Mineral	BFTGOGSKN	143173 (100%)	092F	2006/SEP/28	GOOD		126.343	

**DISCUSSION/INTERPRETATION OF RESULTS RELATIVE TO GEOLOGY
AND CONCLUSION
A.W. Randall P. Eng**

Note:

With respect to the following 4 page report, all references made to Dalmation Resources Ltd. are to be replaced with Perovic Enterprises Inc.

May 25, 2005

Frank Perovic
Dalmation Resources Ltd
5245 Fairmont Street
Vancouver BC
V5R 3V4

**Re: Christina (Tay) Gold Property, Taylor River Area, Vancouver Island
Project Appraisal**

Terms of Reference

Frank Perovic of Dalmation Resources has asked the writer to provide some guidance as to the potential feasibility of locating a successful, economically viable gold mine on the Tay Property. Mr. Perovic has asked for an uncensored appraisal based on the writers own personal observations and conclusions. In Mr. Perovic's words, "...what would you do with this property if you were the owner..." The writer was provided with two summary documents to review as part of this appraisal. The writer visited the property and examined new drill core from a drill program that was underway at the time, in company with Mr. David McLelland (prospector) and the owner representative Mr. Perovic on May 12, 2005

Many of the comments in this document are forward looking statements based on limited evidence from previous field work. The writer agrees to make this appraisal on condition that Dalmation Resources and its officers and representatives accept the suggestions exclusively and entirely at their own risk and indemnify the writer of any and all liability what so ever. A W Randall and RM Resource Management Ltd accepts no responsibility for damages, if any, suffered by any parties as a result of decisions made or actions taken based on this review. The owners, and/or their representatives, in taking receipt of this report, agree to these terms.

The Property

The Dalmation Project area consists of 6 Tenures (claims blocks) including 100 cells and totalling approximately 2106 hectares staked under the new Mineral Titles Online system. All pre-existing claims have been converted to the new system. The property hosts numerous gold and locally a few gold +/- copper-lead-zinc showings. Ten of these showing have been highlighted for additional exploration activity. Only one area, the Tay Vein, has had sufficient exploration drilling on which a modest ore reserve has been developed. This reserve is indicated to be 145,000 tonnes grading 0.063 ounces/tonne (2.16 grams/tonne).

Past Exploration Activity

The Tay mineralization was discovered in 1899 and has had been subject to a number of exploration programs in the intervening years including underground drifting on the Morning Vein, prospecting, geological mapping, a variety of soil and stream sediment geochemical surveys, geophysical surveys including ground surveys IP, Mag and EM as well as a helicopter borne aerial surveys. All this exploration has been thoroughly synthesized and analyzed by Leo Lindinger P. Geo in 1994 and again by A. A. Burgoyne P. Eng in 1996 from which numerous anomalous areas have been identified. Several diamond drilling programs have tested a number of targets over the years with limited success. The most extensive diamond drilling has outlined the Tay Vein with reserves as noted above. All these activities have been well documented in the reports supplied and do not require further discussion here.

Property Assets

1. Wide spread occurrence of gold mineralization which appears to be primarily associated with the Tay-Morning Fissure system
2. Evidence of local concentrations of gold mineralization in shear vein systems, the most well defined (explored) being the Tay Vein.
3. Evidence that the Tay Vein section, which has been the most extensively drilled, is not fully closed off or defined by drilling to date. There is some indication that the deposit may extend particularly to depth but also to the west.
4. Evidence of the potential for locally higher gold grades ranging up to 1 ounce per tonne (30+ grams per tonne).
5. Considerable previous exploration with which to focus any future exploration program.
6. A considerable length of the 6 to 7 kms of favourable shear zone structure has been only minimally explored.
7. Location is close to a very good and easily accessible transportation corridor.
8. Proximity to nearby service centre, bedroom community and workforce.
9. Hydro power right on the doorstep of the property.

Property Liabilities

1. Proximity to the Salmon bearing Taylor River-Sproat Lake fishery system. Any mine development in this area will have to undergo an intensive and extensive environmental assessment before being allowed to proceed. It will have to include extensive measures to capture and clarify sediment laden waters from mining activities and neutralize any acidity in any and all runoff waters emanating from the site as a result of exploration and mining. In addition protective barriers to capture potential fuel, oil and toxic chemical spills would have to be designed and built as part of any mine development.
2. In addition to the above it appears likely that mining will probably have to extend below the Taylor River valley floor. Consequently water is likely to be encountered in underground workings which will have to be pumped out and treated if it is not clean enough to release to the environment as is.

3. The small narrow lensy and discontinuous veins will likely require expensive underground shrinkage stoping methods to mine these deposits, requiring extensive development and which may result in ore losses and/or excessive dilution during mining.
4. The presence of a penstock and turbine system situated very close to the Tay Vein mineralized zone may make it difficult to mine without disturbing this system.

Observations and Recommendations

There is no doubt that this property has potential for discovery of additional gold mineralization. Only a small portion of the total 6 to 7 km strike length of the system has been thoroughly explored. In order to do this properly and in a timely fashion will require a fairly extensive exploration program. The owner's small drilling program, while providing a small amount of valuable information, at the rate it is able to progress would take many years to explore the system.

Geologically the area is situated in rocks which contain alteration and mineralization indicative of the presence of gold +/- base metal ore deposits and with significant structure suitable for channelling of mineralizing fluids and development of locations for ore deposition. In addition the proximity to the Bedwell Batholith is also a possible heat engine and/or structural driver for development and localization of mineralization. The apparent lack of Sicker Formation rocks in the immediate vicinity suggests that a Myra Falls style of massive sulphide deposit is unlikely.

Exploration of this mineral prospect will take considerable exploration talent and substantial funding to make the necessary strides to determine if a viable deposit is present. The owners have made progress along this way with the work they have had done to date, in having all available data analyzed and targets developed.

It is the writer's suggestion that first of all an appraisal of the mineability of the existing deposit be made to develop an idea of the feasibility of mining. For example an estimate of mining method, mining costs and ore recovery potential should be made to ultimately develop a good idea of the required target size for a viable deposit at this location. An examination of the host structures and the expected geometry of ore zones would be helpful in this regard. An appraisal such as this can be made by an ore reserve specialist teamed with a mine development consultant and might be done for under \$30,000 depending on availability of consultants.

Then with an indication of the target size and viability a concerted exploration effort could be undertaken either by optioning the property to a suitable company or the company could raise the money itself on the market and hire a consulting team to manage the work as has already been done. It is estimated that a staged project of \$500,000 to \$1,000,000 should be done to adequately explore this property. A suitable exploration program should include the following elements in the order listed:

1. Review and re-evaluate the targets previously identified focusing primarily on the evidence of gold mineralization and geochemical anomalies. Although there is some indication of correlation to structure and mineralized showings, geophysical surveys in this rugged terrain may be suspect.
2. Do a structural analysis to try to define/confirm the trend of the mineralized zones and to see if the shear zone is bending or is being offset by faulting.
3. Review the work done on the Tay Vein reserve and look to expand the existing resource with additional drilling where the zone appears to be open.
4. Re-examine the underground workings to see if access is safely possible. If accessible sample these underground workings. In addition see if these workings would be a suitable platform for diamond drilling to test other parts of the shear zone system, including possible short extensions of the workings to accommodate this.
5. Reopen existing or develop new roads to access and easily travel to some of the more remote, higher altitude locations on the property. In addition to providing access, these road cuts will provide more exposure for mapping and sampling and the possibility to do some excavator trenching.
6. Do focused mapping, sampling and prospecting along the favourable trend.
7. Begin drilling some of the best targets identified along the mineralized trend.

It would be prudent however before undertaking further work to explore with the Environmental Assessment Office what if any roadblocks (potential "show stoppers") they might see in developing a mine at this location. In addition any concerns with respect to mine development around the small hydro power system should also be explored.

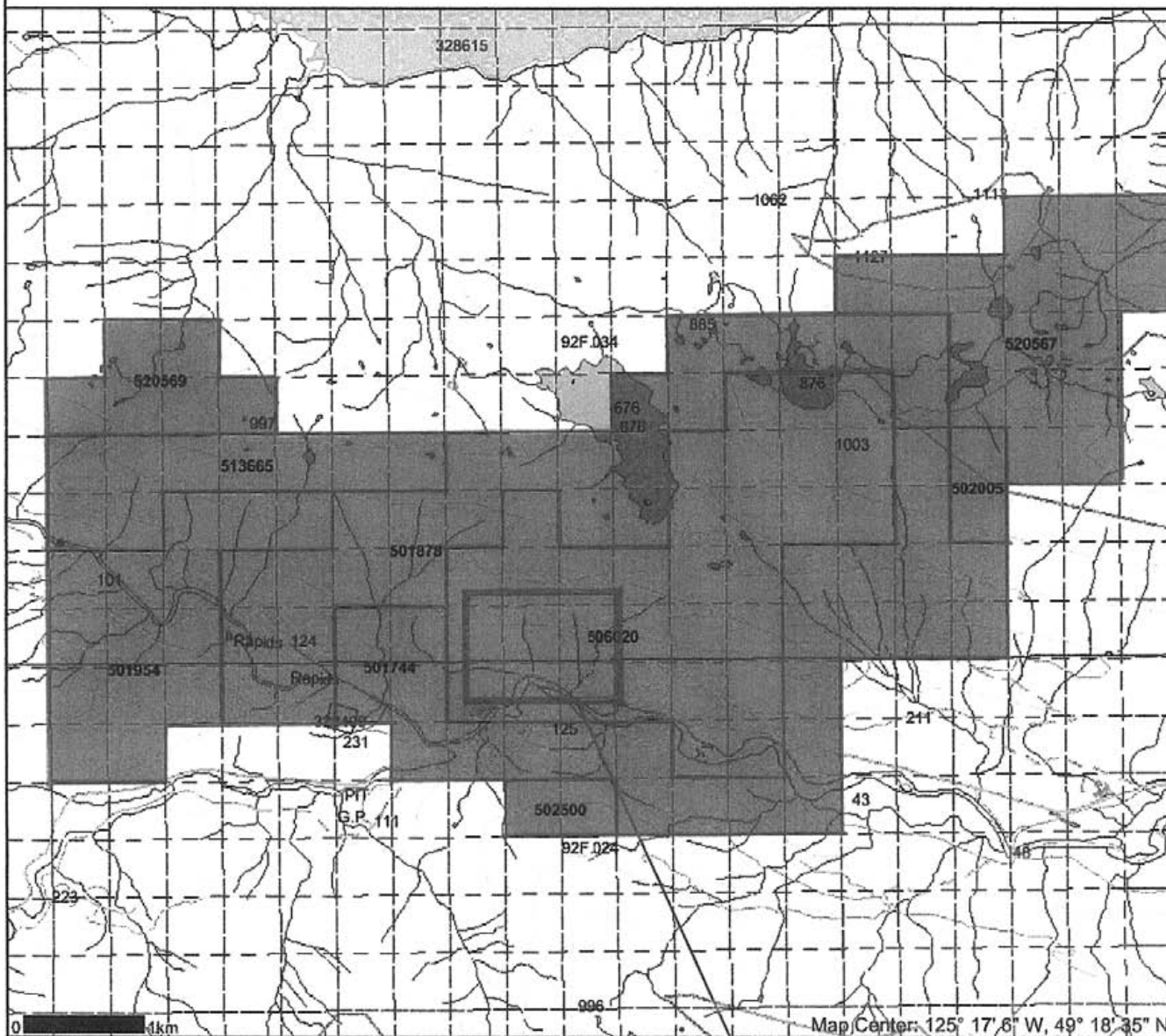
Yours truly,



A. W. Randall P. Eng
RM Resource Management Ltd

Map created Sat Mar 25 23:26:18 PST 2006

Legend



- Indian Reserves
- National Parks
- Parks
- Mineral Titles Grid
- Mineral Tenures
- Reserves (Sites)
- Placer Claim Designation
- Placer Lease Designation
- No Staking Reserve
- Conditional Reserve
- Release Required Reserve
- Surface Restriction
- Recreation Area
- Others
- Mining Divisions
- BCGS Grid
- Contours (1:250K)
- Contour - Index
- Contour - Intermediate
- Area of Exclusion
- Area of Indefinite Contours
- Transportation - Points (TRIM)
- Helipad
- Transportation - Lines (TRIM)
- Airfield
- Airport
- Airstrip
- Airport Abandoned
- Ferry Route
- Road (Gravel Undivided) - 1 Lane
- Road (Gravel Undivided) - 2 Lanes
- Road (Gravel Undivided) - UIC - 1 Lane
- Road (Gravel Undivided) - UIC - 2 Lanes
- Road (Paved Divided) - Not Elevated - 1 Lane Each Way
- Road (Paved Divided) - Not Elevated - 2 Lanes Each Way
- Road (Paved Divided) - UIC - Not Elevated - 2 Lanes Each Way
- Road (Paved Undivided) - Not Elevated - 1 Lane
- Road (Paved Undivided) - Not Elevated - 2 Lanes
- Road (Paved Undivided) - Not Elevated - 4 Lanes
- Road (Paved Undivided) - UIC - Not Elevated - 4 Lanes
- Road (Unimproved)
- Cut (Roadway)
- Embankment/Fill (Roadway)
- Trail
- Bridge - Foot
- Bridge - Trestle
- Tunnel
- Bridge
- Rail Line (Double Track)
- Rail Line (Multiple Track)
- Rail Line (Single Track)

Scale: 1:50,000

DO NOT USE FOR NAVIGATION

Map Center: 125° 17' 6" W, 49° 18' 35" N

Inset (see Figure 5)

Figure 4 Drill Locations related to claim boundary

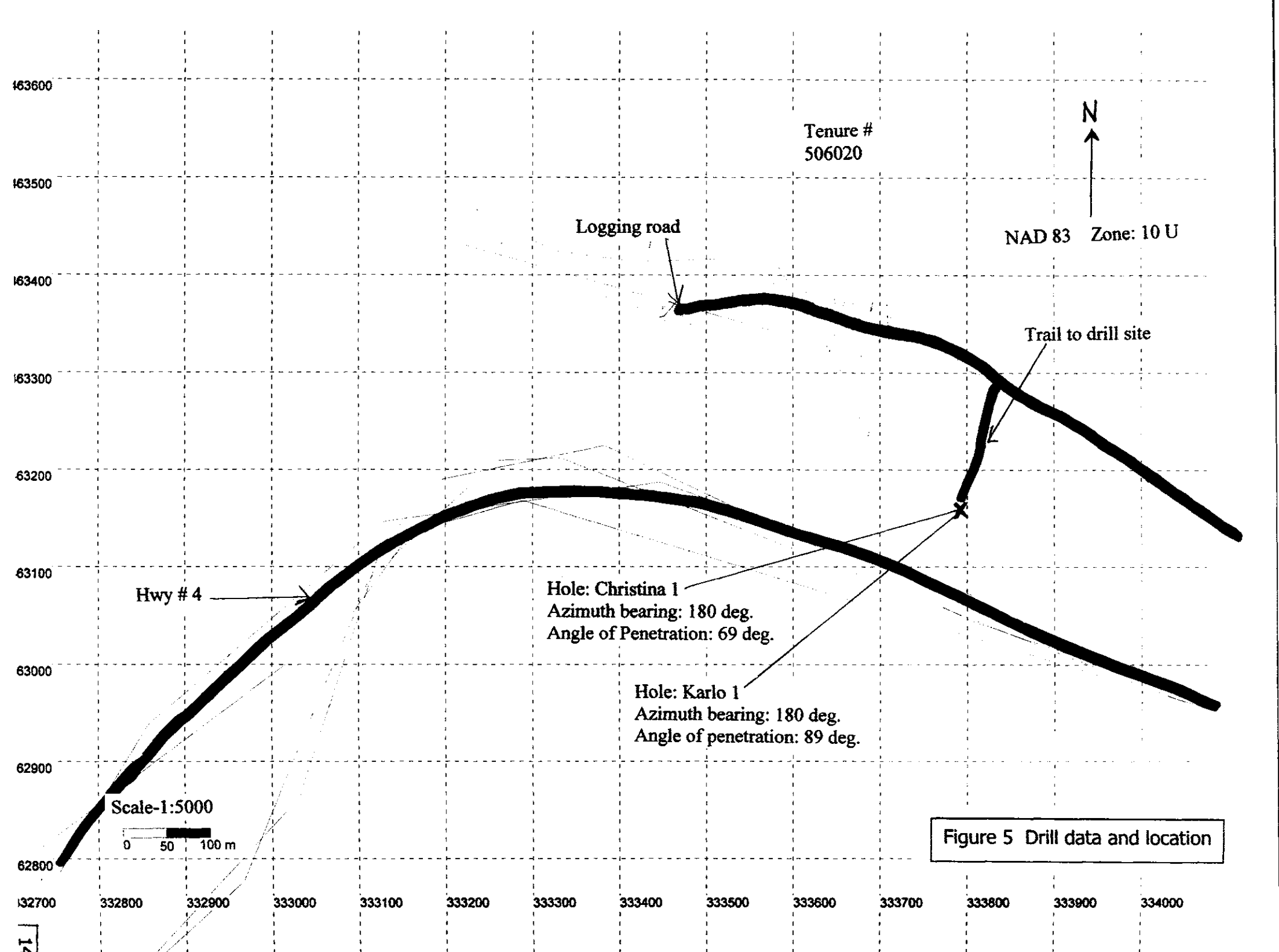
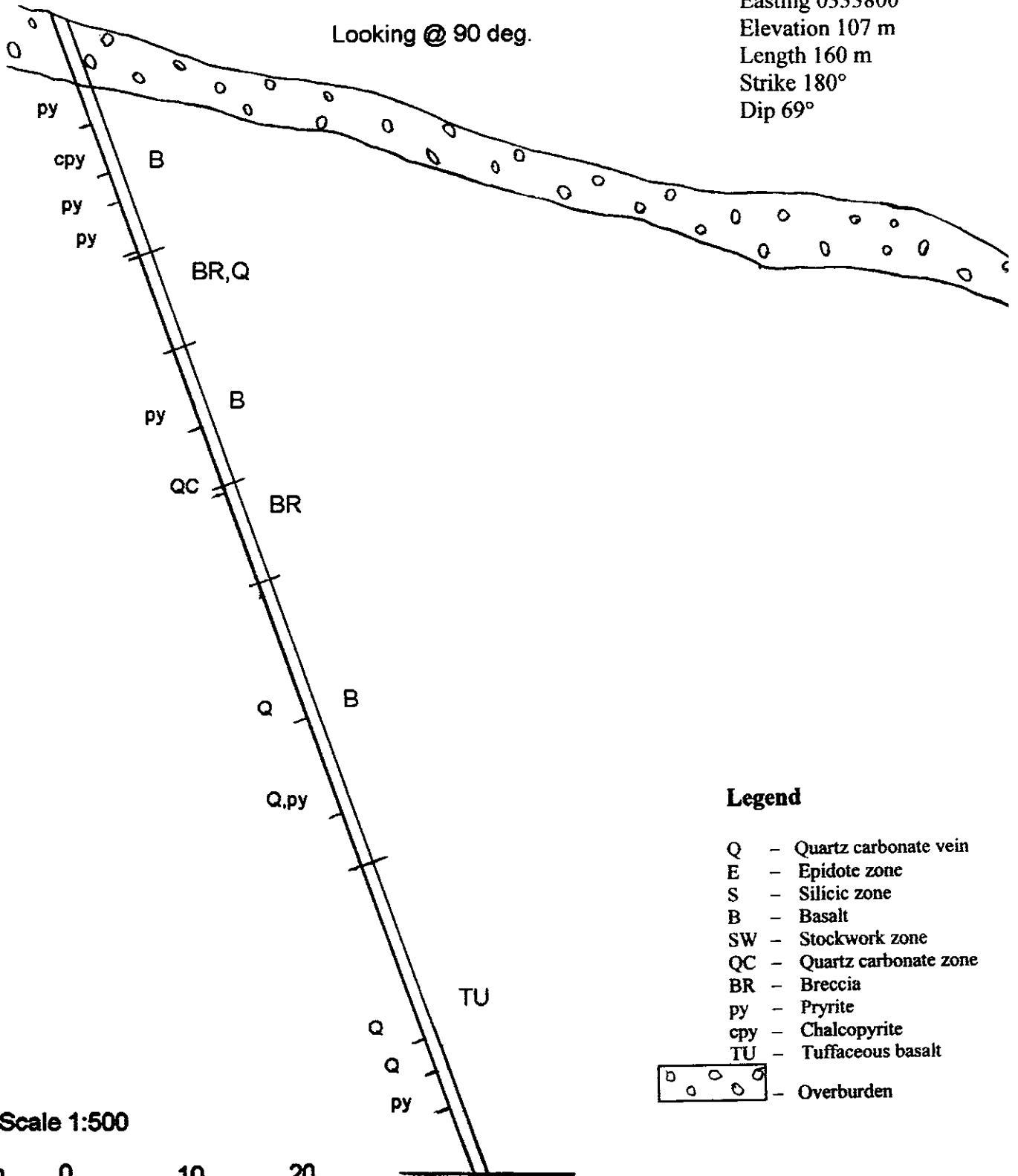


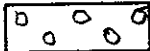
Figure 6 Section DDH Christina 1

Hole Collar
 Northing 5463160
 Easting 0333800
 Elevation 107 m
 Length 160 m
 Strike 180°
 Dip 69°

Looking @ 90 deg.



Legend

- Q - Quartz carbonate vein
- E - Epidote zone
- S - Silicic zone
- B - Basalt
- SW - Stockwork zone
- QC - Quartz carbonate zone
- BR - Breccia
- py - Pyrite
- cpy - Chalcopyrite
- TU - Tuffaceous basalt
-  - Overburden

Scale 1:500

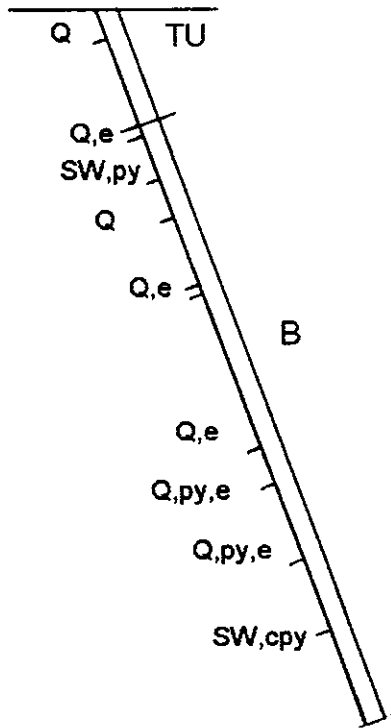


Figure 6 Section DDH Christina 1

Hole Collar
Northing 5463160
Easting 0333800
Elevation 107 m
Length 160 m
Strike 180°
Dip 69°

CONTINUED

Looking @ 90 deg.



Scale 1:500



Figure 7 Section DDH Karlo 1

Hole Collar
 Northing 5463160
 Easting 0333800
 Elevation 107 m
 Length 213.7 m
 Strike 180°
 Dip 89°

Looking @ 90 deg.

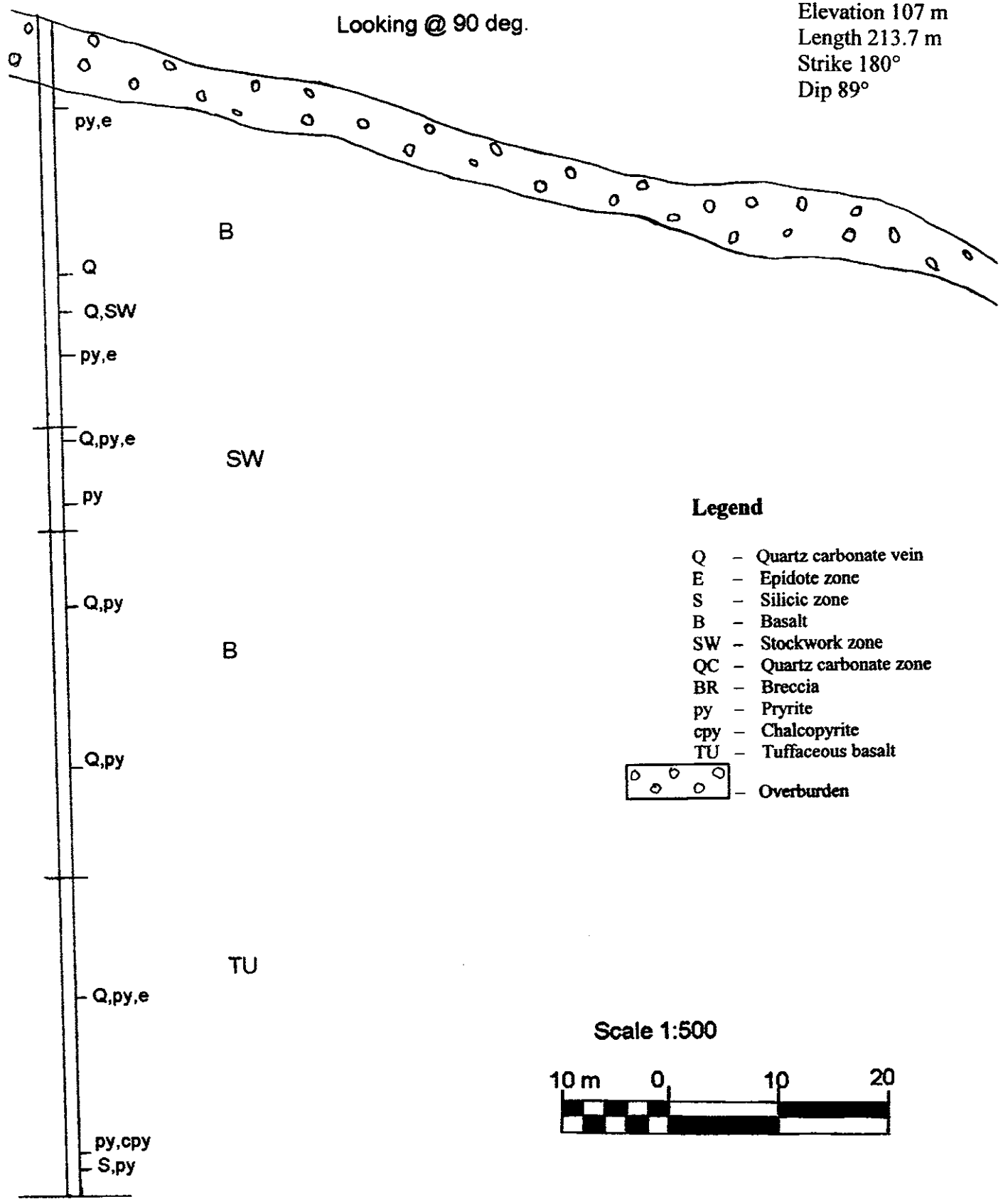
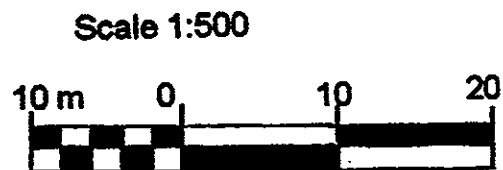
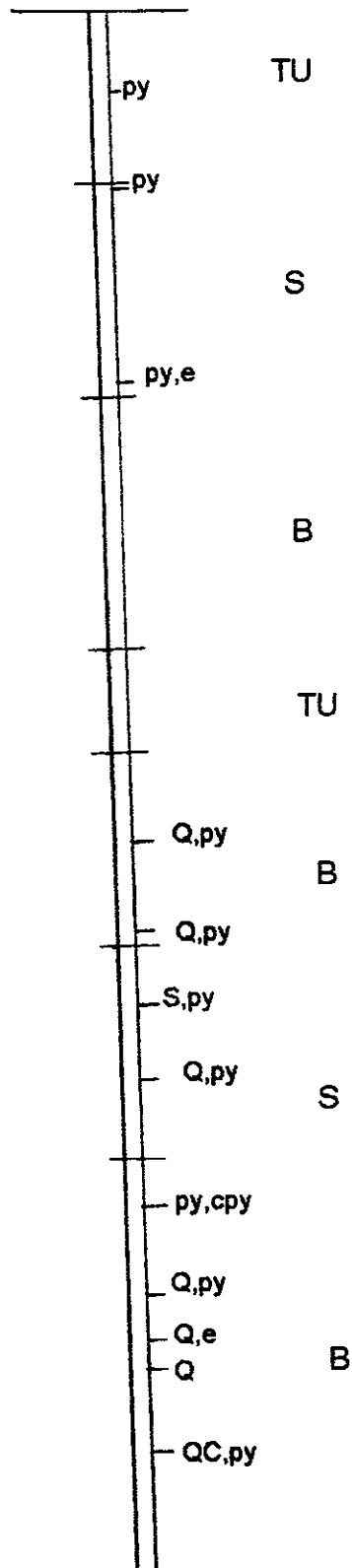


Figure 7 Section DDH Karlo 1

Hole Collar
Northing 5463160
Easting 0333800
Elevation 107 m
Length 213.7 m
Strike 180°
Dip 89°

CONTINUED

Looking @ 90 deg.



COST STATEMENT April 17 - May 30, 2005

Room and Board – 50 man days x \$100.00	\$5,000.00
Chainsaw operating: \$30 x 3	\$90.00
Chainsaw standby: \$10 x 32	\$320.00
Trail work: 1 week – 40 hrs @ \$20/hr	\$800.00
Drilling 1216 feet of BQ (1 7/16) @ \$40 per foot	\$48,640.00
Core rack, storage, material, plus labour @ \$20/hr x 15 hrs	\$600.00
Lab Costs	\$575.39
Project Supervision/management @ \$30/hr x 300	\$9,000.00
Logistical support @ \$20/hr x 100 hrs	<u>\$2,000.00</u>
Subtotal	\$67,025.39
Pick-up Truck 4x4 - \$940/month + \$230/wk	\$1,170.00
Gas/ferry trips	<u>\$3,244.97</u>
Subtotal	\$4,414.97
Total Expenditures	<u>\$71,440.36</u>

COST STATEMENT May 31, 2005 - September 27, 2005

Room and Board	\$234.55
Reclamation & clean up work (\$20/hr + \$30/hr) x 20 hrs	\$1,000.00
Core Storage	470.80
Management \$30/hr x 8hrs (Consultation visit/analysis with Regional Geologist)	\$240.00
Geological Consultant	<u>\$1,019.18</u>
Subtotal	\$2,964.53
Pick-up Truck 4x4 - \$50/day x 6 days	\$300.00
Gas/Ferry Trips	<u>\$441.71</u>
Subtotal	\$741.71
Total Expenditures	<u>\$3,706.24</u>

COST STATEMENT September 28, 2005 - April 10, 2006

Room and Board	\$170.89
Supplies	45.57
Core Logging	\$900.00
Core Storage	706.20
Geological Consultant	\$505.55
Report Preparation	<u>\$1,200.00</u>
Subtotal	\$3,528.21
Pick-up Truck 4x4 - \$50/day X 2 days	\$100.00
Gas/Ferry Trips	<u>\$495.33</u>
Subtotal	\$595.33
Total Expenditures	<u>\$4,123.54</u>
GRAND TOTAL	<u>\$79,270.14</u>

Persons whom worked on Tay-Christina Project

Claude Lessard	--	Diamond driller
Shane Hess	--	General labourer
Gord Corcoran	--	General labourer
Frank Perovic	--	Project Manager
Christina Perovic	--	Logistical Support/labourer
Alf Randall, P. Eng	--	Consultant

STATEMENT OF QUALIFICATIONS

I, Franjo Perovic, of 8456 Karr Place, Delta, BC do hereby certify that:

- 1) I hold a valid free miners certificate since 2000.
- 2) I completed the Prospecting/Exploration Method program hosted by BCIT and BC/Yukon Chamber of Mines in 2002.
- 3) I have completed the Prospecting Field School hosted by BCIT and BC/Yukon Chamber of Mines in 2003.
- 4) I have prospected independently since 2000.
- 5) I currently own 100% interest in the subject property.

Dated at Delta, BC, on April 10, 2006.



Franjo Perovic

LIST OF REFERENCES CITED IN THIS REPORT:

Dalmation Resources Ltd. Report on Tay Gold Property
Dec. 1983 ▪ V. Cukor, P. Eng. ▪ NVC Engineering Ltd.
Figure 4 (OUTCROP GEOLOGY PLAN)

Dalmation Resources Ltd. Report on 1994 phase 1 diamond drill program on the
Tay Main (East) zone and Slide zone with a summary of economic potential on
the Tay Property

Leo J. Lindinger, P. Geo. ▪ July 20, 1994

Figure 4 (TAY AREA COMPILATION MAP)

LIST OF SOFTWARE PROGRAMS used in support of the exploration /
development and preparation of this report:

- Microsoft Windows XP, MS Word, MS Excel
- GPS Trackmaker

Drilling Project April - May 2005, Vancouver Island by Frank Perovic

Hole: Christina 1, April 2005 99% Recovery
NAD 83 UTM co-ordinate
Zone 10U 5463160 0333800

Azimuth bearing 180° No dip tests were performed
Angle of Penetration 69°

0 – 20'	casing
17' – 49'	Basalt (fine grained, dark green, magnetite present in rock). Quartz carbonate stringer veinlets and minor sulphides with blebs as big as 3mm diameter occurring
49' -50'	Sample 1: Calcite with green, pink epidote and minor sulphides (mostly pyrite)
50' – 74'	Basaltic to andesitic with stringer veins up to 1cm wide, minor pyrite, chalcopyrite
74' – 75'	Part of Sample 2: Light greenish gray
75' – 105'	Part of Sample 2: Crackle brecciated, partially healed basaltic to andesitic. Quartz carbonate veinlets with minor sulphides .
105' – 129.5'	Dark green basalt. Mylonitized (chert like) Banded section-purple with 6 bands across a 2cm fracture. Also present are pieces of mica like white fine grained scales and flakes (sericite) found in fractures.
129.5' – 131'	Sample 3: Minor disseminated sulphides (mostly pyrite).
131' – 145.9'	Greenish gray basalt with hairline fractures to dark green.
145.9' – 147'	Sample 4: Dark green basalt with 1/3 bleb of quartz carbonate
147' – 176'	Dark green basalt 163' – 176' heavily fractured/brecciated quartz calcite
176' – 178'	Sample 5: Partially healed fracturing and brecciation with minimal sulphide content
178' – 180.4'	Dark green to light green basalt.
180.4' – 181.2'	Sample 6: Veinlets of quartz-calcite within basalt.
181.2' – 267.2'	Dark green to light green basalt. Fractures from hairline to 2cm thick stringer veinlets of quartz carbonate, minor pyrited areas.
267.2' – 268.5'	Sample 7: Fine grained, dark grayish green explosive type volcanic (tuffaceous)
268.5' – 328.2'	Fine grained tuffaceous To large pyroclasts with quartz-carbonate veinlets up to 1 cm wide.
328.2' – 330'	Sample 8: Fine grained, greenish gray tuffaceous (volcanic – basalt) with 4 mm diameter pyroclasts, minor sulphides.
330' – 341'	Tuffaceous with pyroclasts as large as 6mm diameter.
341.2' – 342'	Sample 9: Tuffaceous with stringer veinlets 1 cm wide.
342' – 389.4'	Tuffaceous to 357' and becoming fine grained basalt again. Small veinlets, 1- 5" wide vein 380 – 383' tuffaceous
389.4' – 390.5'	Sample 10: 2' wide intersection of quartz-calcite with green and pink epidote.
390.5' – 398'	Basalt with intersections of stringer veinlets up to 2cm wide.
398' – 400.2'	Sample 11: Fine-grained basalt with minor pyrite and hairline stringer stockwork veinlets.
400.2' – 427.2'	Basalt with banded intersecting veinlets. 5' intersection of quartz-calcite vein and numerous stringers as throughout entirety of hole.
427.2' – 428.3'	Sample 12: Quartz carbonate intersection with green and pink epidote.
428.3' – 463.5'	Basalt with hairline stringers with a 3" wide veinlet.
463.5' – 465'	Sample 13: 2 - 1 cm wide veinlets and 1 - 4" wide intersection of quartz with green and pink epidote. Remainder is basalt.
465' – 470.5'	Fine grained dark gray-greenish basalt.

- 470.5' – 471.7' Sample 14: 11" intersection of quartz with pyrite associated with dark green and pink epidote.
- 471.7' – 480' Basalt
- 480' – 481' Sample 15: Basalt (lighter gray with light green blebby composition). Some fizzle – carbonization
- 481' – 503' Basalt with 1 intersection of 14" quartz with minor green epidate and minor visible sulphides.
- 503' – 505' Sample 16: Basalt with stockwork veinlets and blebbing of chalcopyrite found.
- 505' – 515' Basalt with a couple of 2" wide blotches of quartz.

End of hole.

Drilling Project April - May 2005, Vancouver Island by Frank Perovic

Hole: Karlo 1, May 2005
NAD 83 UTM co-ordinate
Zone 10U 5463160 0333800

99% Recovery

Azimuth bearing 180°
Angle of Penetration 89°

No dip tests were performed

0 – 20'	casing
10' – 28'	Basalt- fine grained
28' – 28.7'	Green and pink epidote with minor sulphides (pyrite).
28.7' – 78'	Basalt with stringers as large as 15 mm.
78' – 90'	Stringer calcite-quartz filled fractures (hairline to 15mm and blotches)
90' – 91'	Sample 17: 7.5" of quartz-calcite green epidote, some pink epidote with minor sulphides.
91' – 102'	Several stringers and blotches of quartz-calcite
102' – 104'	Sample 18: Stockwork veinlets and blotchy intersections of quartz calcite. Epidote (green and some pink) with minor sulphides.
104' – 126'	Fine grained grayish green basalt with stringers throughout (quartz-calcite). Green epidote throughout
126' – 158'	Most areas covered by stockwork stringers and blotches of quartz calcite. Presence of green epidote with minor sulphides.
158' – 171'	Fine grained basalt with web like stringers.
174' – 265'	Basalt mostly intense to mild fractures filled with quartz carbonate turning to a light greenish gray. Some small pyritized blebs 1 mm diameter.
265' – 345'	Tuffaceous basalt with presence of pyrite blebs of up to 3mm in diameter, fractures are filled with stringers of quartz-calcite with minor green and pink epidote.
344.5' – 345'	Sample 19: Tuffaceous basalt with a greater presence of sulphide blebs.
345' – 394'	Tuffaceous basalt fractured, silicified and carbonatized stringers with minor sulphides.
394' – 448'	Heavy fracturing – Silicification, carbonatization (shear zone unhealed) with minor sulphides containing 7.5 mm blebs occurring in basalt.
448' – 502'	6 blotches of quartz-calcite with epidote (green and pink) all are approximately 3 – 5" wide in basalt.
502' – 528'	Basalt becomes lighter gray tuffaceous with greater sulphide content.
528' – 563'	Basalt, fine grained, minor sulphides often associated with blebs of green epidote.
563' – 568'	Intense fracturing filled with a quartz carbonate along with minor sulphide content (blebs of sulphide as large as 8 mm diameter).
568' – 611'	579' – 580.3' heavily fractured, silicified/carbonatized. 581' – 583' also quartz carbonate with minor sulphide content in fine grained basalt (dark greenish gray to lighter greenish gray). Stringer veinlets throughout up to 1 cm thick.
611' – 638'	Heavily fractured zone filled with quartz carbonate and minor sulphides (mainly pyrite has been seen throughout but chalcopyrite and arsenopyrite have been present)
638' – 654'	Fine grained basalt with mainly minor pyrite blebs. Stringers of quartz-calcite up to 7 mm wide.
654' – 655'	Quartz-calcite area (pink hematite staining and green epidote)
655' – 659'	Fine grained basalt dark greenish gray with stringers of quartz-calcite.

659' - 664'

blotches and veinlets of quartz-calcite.

664' - 701'

@ 685', 693', 694', 695', 697" - 2", 7", 2", 2", 3" respectively. Blotches of quartz-calcite with minimal sulphide content in basalt (fine grained dark greenish to gray).

End of hole.

APPENDIX A

Cross Reference of Sample Numbers

Drill Core Log Sample #	Assay Tag/Sample #
1	B085253
2	B085254
3	B085255
4	B085256
5	B085257
6	B085258
7	B085259
8	B085260
9	B085261
10	B085262
11	B085263
12	B085264
13	B085265
14	B085266
15	B085267
16	B085268
17	B085251
18	B085252
19	B085270

Note: other assays shown are not related.



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To: PEROVIC, FRANK
8456 KARR PLACE
DELTA BC V4C 3X7

Page: 2 - A

Total # Pages: 2 (A - C)

Finalized Date: 23-MAY-2005

Account: PERFRA

CERTIFICATE OF ANALYSIS VA05038524

Sample Description	Method Analyte Units LOR	WEI-21	AU-AA23	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41
		Recvd Wt. kg	Au ppm	Ag ppm	Al %	As ppm	B ppm	Ba ppm	Be ppm	Bi ppm	Ca %	Cd ppm	Co ppm	Cr ppm	Cu ppm	Fe %
		0.02	0.005	0.2	0.01	2	10	10	0.5	2	0.01	0.5	1	1	1	0.01
B085251		0.38	0.009	<0.2	1.72	31	<10	10	<0.5	<2	6.06	<0.5	13	45	119	2.87
B085252		0.56	0.011	<0.2	3.35	12	<10	10	<0.5	<2	14.5	<0.5	25	63	112	5.07
B085253		0.56	0.009	0.2	2.57	2	10	<10	<0.5	<2	4.39	<0.5	13	37	204	2.48
B085254		0.48	0.011	0.2	4.74	6	10	20	<0.5	<2	5.94	<0.5	26	66	112	5.62
B085255		0.48	0.007	<0.2	5.21	7	<10	<10	<0.5	<2	10.05	<0.5	31	96	78	7.61
B085256		0.50	0.006	0.2	2.83	<2	<10	10	<0.5	<2	2.32	<0.5	21	57	160	4.64
B085257		0.74	0.011	<0.2	4.40	<2	10	10	<0.5	<2	8.43	<0.5	33	85	253	6.43
B085258		0.22	<0.005	<0.2	3.49	<2	<10	<10	<0.5	<2	9.78	<0.5	28	76	173	6.50
B085259		0.62	0.009	<0.2	3.32	<2	<10	20	<0.5	<2	2.63	0.6	10	26	196	2.72
B085260		0.80	0.017	<0.2	3.12	<2	<10	50	<0.5	<2	3.04	<0.5	14	24	83	3.84



GEOCHEMICAL ANALYSIS CERTIFICATE

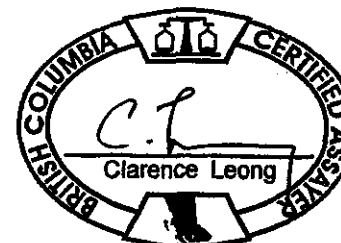


Perovic, Frank File # A502259
8456 Kerr Pl., Delta BC V4C 3X7 Submitted by: Frank Perovic

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppb	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	Hg ppm	Sc ppm	Tl ppm	S %	Ga ppm	Se ppm
085261	1.8	114.4	.6	65	.1	47.8	25.9	1025	5.43	1.2	.1	13.1	.4	29	<.1	.1	<.1	168	4.07	.068	3	61.8	2.43	20	.202	5	4.28	.046	.03	<.1	.01	16.0	<.1	.06	11	<.5
085262	.4	122.5	.4	15	.1	19.4	9.1	269	1.60	4.0	.1	9.3	.1	34	.2	.3	<.1	52	10.24	.015	<.1	18.5	.28	2	.106	42	1.24	.020	<.01	.1	.01	2.4	<.1	.10	6	<.5
085263	.9	174.8	.5	45	.1	41.3	18.1	479	3.36	1.5	.1	7.4	.3	36	.1	.1	<.1	103	2.06	.044	2	53.5	1.23	11	.335	5	2.19	.182	.04	<.1	<.01	7.2	<.1	<.05	6	<.5
085264	.6	64.9	.5	23	<.1	23.3	10.5	290	2.00	2.1	<.1	18.9	.1	25	.1	.1	<.1	64	6.74	.024	1	32.1	.58	2	.215	4	2.15	.042	.01	.1	.01	3.6	<.1	<.05	9	<.5
085265	.4	163.9	.4	34	.1	32.4	13.8	295	2.43	1.7	<.1	5.7	.2	38	.1	.1	<.1	73	2.38	.030	1	40.3	.95	7	.252	76	2.20	.168	.03	<.1	.01	5.3	<.1	<.05	6	<.5
085266	2.3	237.4	.4	24	.1	24.5	10.0	281	1.97	6.9	<.1	10.7	.1	21	.3	.1	<.1	58	1.45	.027	1	40.3	.68	7	.252	19	1.25	.104	.03	<.1	<.01	4.6	<.1	<.05	4	<.5
085267	.5	129.0	.8	51	.1	35.7	18.9	720	4.03	2.7	.1	3.3	.3	54	.1	.1	<.1	143	5.78	.054	3	56.6	1.39	14	.411	5	2.40	.121	.05	.1	<.01	13.3	<.1	.11	9	<.5
085268	.9	183.0	.6	61	.1	46.8	23.0	617	4.48	4.2	.1	5.2	.2	23	.2	.2	<.1	123	2.72	.041	2	75.3	2.04	5	.414	9	3.08	.104	.02	.1	.01	8.9	<.1	.06	11	<.5
085270	.7	100.8	.6	39	.1	13.4	19.8	417	3.75	3.5	<.1	4.7	.4	119	<.1	.1	.1	115	3.08	.082	2	20.2	1.09	72	.139	2	3.89	.398	.06	<.1	<.01	3.4	<.1	.75	8	.7
085271	.8	191.9	1.8	59	.2	55.2	28.7	601	5.58	6.0	.1	21.7	.3	22	.1	.2	<.1	198	3.61	.065	3	106.7	1.54	4	.464	9	3.33	.055	.01	.2	.08	10.1	<.1	1.58	10	.6
STANDARD DS6	11.7	124.7	29.7	145	.3	24.4	10.4	710	2.89	21.8	6.4	48.3	3.1	40	6.2	3.5	5.2	55	.85	.080	14	190.5	.59	163	.077	17	1.91	.075	.16	3.7	.23	3.2	1.7	<.05	6	4.6

GROUP 1DX - 15.00 GM SAMPLE LEACHED WITH 90 ML 2-2-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR, DILUTED TO 300 ML, ANALYSED BY ICP-MS.
(>) CONCENTRATION EXCEEDS UPPER LIMITS. SOME MINERALS MAY BE PARTIALLY ATTACKED. REFRACTORY AND GRAPHITIC SAMPLES CAN LIMIT AU SOLUBILITY.
- SAMPLE TYPE: Core R150

Data FA _____ DATE RECEIVED: MAY 30 2005 DATE REPORT MAILED: June 8/05





GEOCHEMICAL ANALYSIS CERTIFICATE



Perovic, Frank File # A502260

8456 Karr Pl., Delta BC V4C 3X7 Submitted by: Frank Perovic

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppb	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	Hg ppm	Sc ppm	Tl ppm	S %	Ga ppm	Se ppm
085269 STANDARD	5.0	14.0	8.2	15	.3	5.2	9.0	94	3.51	2476.4	<.1	528.8	.3	8	.2	2.1	<.1	26	.09	.050	2	10.4	.32	25	.002	2	.80	.005	.27	<.1	<.01	2.2	<.1	1.83	2	1.5
	11.7	124.7	29.7	145	.3	24.4	10.4	710	2.89	21.8	6.4	50.8	3.1	40	6.2	3.5	5.2	55	.85	.080	14	190.5	.59	163	.077	17	1.91	.075	.16	3.7	.23	3.2	1.7	<.05	6	4.6

Standard is STANDARD DS6.

GROUP 1DX - 15.00 GM SAMPLE LEACHED WITH 90 ML 2-2-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR, DILUTED TO 300 ML, ANALYSED BY ICP-MS.

(>) CONCENTRATION EXCEEDS UPPER LIMITS. SOME MINERALS MAY BE PARTIALLY ATTACKED. REFRACTORY AND GRAPHITIC SAMPLES CAN LIMIT AU SOLUBILITY.

- SAMPLE TYPE: Rock R150

Data N FA _____

DATE RECEIVED: MAY 30 2005

DATE REPORT MAILED: June 7/05

