

...CYANIDE PROCESS...

By Gerald Voss Hopkins, M. E.

Mr. Gerald Hopkins, M. E., the metallurgist in charge of the cyaniding plant at Silica, owned by the British Columbia Ballion Extraction company, address before the School of Mines on Tuesday evening. The full text of the address is as follows:

In testing gold ores with a view to their practical treatment by any particular process the main object is to arrive at a decision: First, as to whether that particular process is the most suitable and economical one for their treatment, and secondly, as to the best way to apply the process to secure the greatest profit from their treatment by it.

In the following notes I have endeavored to give an outline of a course of testing in connection with the cyanide process, applicable to the general run of ores, etc., but without attempting to meet the requirements of really exceptional cases, also a few remarks on the treatment of ores from a work standpoint.

Whatever the material to be tested happens to be, before proceeding to the actual testing, it is of the essential importance to find out whether it contains sufficient gold or silver to pay the cost of treatment at all. No hard and fast line can be drawn in regard to minimum gold contents which may pay for treatment, this depends so much on the character of the material to be treated, and here I may mention that there are cases where ores containing little more than \$2 in gold are treated by the cyanide process direct and yield a fair margin of profits.

If assays have shown the material to contain a reasonable amount of gold, the next step is to carry out a preliminary test, of which a determination of the quantity of acidity, if any, and the consumption of cyanide during short contact of the material with weak cyanide solution. With or without the addition of an alkali treatment according to results of acidity tests, are the most important—for the former 200 grammes of ore is made into pulp with a little water and the pulp tested, say with blue litmus paper; if this is reddened, soluble acidity is present and a weak standard soda solution is run in until neutral point is arrived at. After this, however, though the blue litmus is no longer red, the red litmus may not be turned blue, by a slight further addition of soda solution, and in this case, soda is again run in until the pulp does turn blue. This second quantity of soda is a measure that goes by the name of "latent acidity," and is due to the presence of various constituents which, though not strong acids themselves, have the power of combining with and neutralizing alkali.

If the material is found to be free from acidity, or only slightly so, 200 grammes are placed in a bottle with similar quantity of weak cyanide solution and shaken for half an hour or so, and the cyanide remaining undecomposed determined; if this is not excessive the ore is generally passed as fit for cyanide treatment without any previous treatment.

If, on the other hand, acidity was found and the consumption of K. Cy. exceeded 0.15 per cent on the weight of ore (i.e., viz. 3 lb. K. Cy. per ton of ore) it is necessary to subject the material to preliminary treatment in order to remove as much as possible of the cyanides present.

From these preliminary trials, one can generally decide whether the material is fit for direct treatment or whether it requires preliminary treatment to remove as much as possible of the cyanides present. First—Class of ore, etc., suitable for direct treatment, from which the gold is readily extracted by cyanide solutions with moderate cyanide consumption in reasonable time, and which are not exceptionally clayey, etc.

Secondly—Class of ore, etc., which are not suitable for direct treatment, but which require preliminary treatment—From which the cyanide process, if applied direct, under normal conditions, does not effect satisfactory extraction of the gold, or in the treatment of which the consumption of cyanide is abnormal, or which are exceptionally clayey, etc.

Having broadly classified the ores, etc., the following is an outline of the methods usually adopted for their examination: The material belonging to the first class is crushed to several times the fineness of a part through a 30-mesh sieve—a part through a 40 and a part through a 60. A series of percolation tests are then carried out on the various portions, using cyanide solutions of selected strength. The usual strength for this class of material is from 0.1 to 0.5 per cent and the quantity employed varies from 1.3 to one part for each part of the material to be treated.

The course of the experiments is followed by assaying an average sample of the material as charged into the percolator, and from time to time assaying and testing the solution, the percolator may be by upward and downward flow or entirely by downward flow. For small trials, however, especially on dry materials, it is best that the first percolation be carried out in a wide shallow dish, the remainder of the trial, as by this method channeling is best avoided and a uniform flow of solution over the particles more or less insured.

The normal time of treatment is three to four days, but if the extraction is not fairly complete in that time it is prolonged until such is the case.

Sometimes a better extraction is obtained by a double treatment, i. e., after the treatment has proceeded for a certain length of time, the residues are washed, discharged, recharged to the percolator and treated a second time with fresh solution.

The apparatus required for making these tests is of the simplest description. A large lamp glass, covered with a perforated rubber cork, fitted with a piece of glass tube, which is further lengthened by two or three inches of black rubber tube, which in turn is directly partially closed by means of a screw clip.

The lamp glass is provided with a false bottom of coarse quality, covered with coarse tannin serves very well, or a sound barrel cut in half would do, one of which as the percolator is fitted with a false spar bottom covered with a filter cloth; and the other as the receiver, may be smaller without a false bottom. The percolator is fitted with a hole in the bottom under the false bottom, this is fitted with a rubber bung carrying a piece of glass or iron tube, to which is attached rubber tubing, provided with a screw clip. For upward percolation the tubes of the two vats are connected by a piece of glass tubing and the method of using the apparatus is obvious.

Sometimes a better extraction is obtained by a double treatment, i. e., after treatment has proceeded for a certain length of time, the residues are washed, discharged, recharged to the percolator, and treated a second time with fresh solution.

If the material under examination consists of tailings, etc., it is advisable to try in the first instance to obtain satisfactory extraction from it in the condition received.

ed, so as to avoid regrinding; if this is impossible the material should be graded and the various gradings assayed to find the gold, as frequently the gold is found to be locked up in the coarser particles of the residue; on these being finely ground, a further trial often proves satisfactory.

If the material is rich—such as concentrates—the treatment, if by percolation, is often prolonged to a month or so. It is sometimes possible to obtain a better material by the quicker agitation method, but if percolation, even if prolonged, gives a good extraction, it is usually found more economical.

Experiments carried on as above will give sufficient data for a decision as to the best general conditions under which their treatment in practice.

Second—Class of ore not suitable to direct cyanide treatment, but requiring preliminary treatment require a more searching investigation. It is impossible to enumerate the possible causes of the material's tendency to cause direct treatment of them by cyanide unsatisfactory. The following are, however, those most commonly met with:

A. A considerable portion of the gold exists irregularly distributed throughout the materials as comparatively large particles, or in case of tailings, large particles of amalgam are left in them.

B. Some of the constituents of the material exist in a form which is readily attacked by cyanide, causing abnormal consumption and simultaneously with this, soluble sulphides are formed, which seriously retard the solution of the gold and silver.

C. A considerable part of the gold exists in combination with antimony, tellurium, bismuth, etc.

D. The material is clayey or slimy, so as to prevent or seriously retard the filtration of the solution through it.

Class A—Ores, etc., of this character are first treated with mercury, to remove the coating of antimony, tellurium, etc., of the well-known methods, or subjected to concentration. This is often adopted in treating tailings, and in this case tables are used. The residues are then treated either by percolation or agitation method of applying cyanide, and if the only difficulty to direct treatment comes under this heading, the effect of double treatment is usually satisfactory.

In addition to the above treatment the combined treatments can be carried out simultaneously with mercury or amalgamated copper plates and cyanide.

In case of float gold it is sometimes found preferable to float the cyanide and the residue by amalgamation methods.

Class B—Apart from the abnormal consumption of cyanide caused by the materials being acid, a considerable arising from the material, the cause of the abnormal consumption of cyanide is generally prevented by preliminary alkaline treatment, the usual trouble is the presence of certain constituents of copper, iron, etc., and the physical condition in which they exist.

Abnormal consumption of cyanide is usually coupled with poor extraction and the formation of soluble sulphides. The method of determining the cause of the cyanide consumption as usually employed is to examine the solutions: (1) for cyanogen compound; (2) carbonates, oxalates and sulphides; (3) the bases in solution; (4) sulphides in solution; (5) alkalinity.

The general methods of procedure with ores containing cyanides is to subject the material to a water washing which will remove the soluble cyanides, followed by a washing of dilute caustic soda to eliminate basic salts of iron. If this is not efficacious free washing, stimulating Spitzglaster treatment, floating off the bulk of the oxides and hydrates which are present; by this method the light pyritic slimes are also removed which will often considerably reduce the cyanide consumption. Time process, if applied direct, under normal conditions, does not effect satisfactory extraction of the gold, or in the treatment of which the consumption of cyanide is abnormal, or which are exceptionally clayey, etc.

Having broadly classified the ores, etc., the following is an outline of the methods usually adopted for their examination: The material belonging to the first class is crushed to several times the fineness of a part through a 30-mesh sieve—a part through a 40 and a part through a 60. A series of percolation tests are then carried out on the various portions, using cyanide solutions of selected strength. The usual strength for this class of material is from 0.1 to 0.5 per cent and the quantity employed varies from 1.3 to one part for each part of the material to be treated.

The course of the experiments is followed by assaying an average sample of the material as charged into the percolator, and from time to time assaying and testing the solution, the percolator may be by upward and downward flow or entirely by downward flow. For small trials, however, especially on dry materials, it is best that the first percolation be carried out in a wide shallow dish, the remainder of the trial, as by this method channeling is best avoided and a uniform flow of solution over the particles more or less insured.

The normal time of treatment is three to four days, but if the extraction is not fairly complete in that time it is prolonged until such is the case.

Sometimes a better extraction is obtained by a double treatment, i. e., after the treatment has proceeded for a certain length of time, the residues are washed, discharged, recharged to the percolator and treated a second time with fresh solution.

If the material under examination consists of tailings, etc., it is advisable to try in the first instance to obtain satisfactory extraction from it in the condition received.

ed, so as to avoid regrinding; if this is impossible the material should be graded and the various gradings assayed to find the gold, as frequently the gold is found to be locked up in the coarser particles of the residue; on these being finely ground, a further trial often proves satisfactory.

If the material is rich—such as concentrates—the treatment, if by percolation, is often prolonged to a month or so. It is sometimes possible to obtain a better material by the quicker agitation method, but if percolation, even if prolonged, gives a good extraction, it is usually found more economical.

Experiments carried on as above will give sufficient data for a decision as to the best general conditions under which their treatment in practice.

to provide sufficient percolators to carry out operations in accordance with the method laid down as result of previous investigations. Thus, if necessary to percolate and wash for two days and a half and half a day is occupied in filling and emptying the percolators, it is obvious that one vat can only have a fresh charge once in three days, therefore a 50-ton per day plant will require three percolators of 50-ton capacity and an additional one for contingencies.

Agitators: These, if used, are worked on day and night shifts and are thus capable of treating three charges per diem of 1.3 ton agitator charges, thus a 50-ton per day agitator plant running on material requiring six hours agitation would comprise seven agitators of about three-ton capacity, treating three charges per day and three 50-ton percolators.

Reservoirs: These should be of sufficient capacity to contain at one time all the solutions, strong and weak, in circulation in the works. As a general rule reservoirs accommodating 1,000 to 1,500 gal. one-third of the cubic contents of the tanks.

Extractors: These are long boxes or troughs containing a prepared fine and so arranged that the gold cyanide solution is forced by baffle boards to pass before it leaves the tank, therefore a 50-ton per day plant will require three extractors of 50-ton capacity and an additional one for contingencies.

They are generally constructed with 10 zinc compartments, each of such dimensions that the amount of solution to be treated may flow up through the zinc easily and steadily.

In practice, each compartment is designed to hold about one cubic inch of zinc for every ton per month treated. Thus a 50-ton plant treating 500 tons per month would have extractors with compartments of such size as to contain about 500 cubic inches of prepared zinc.

This is a vast improvement on the old method of using a six-inch centrifugal wheel well on such a plant.

Slimes: These are removed by means of a slimes pump, which is used to collect the slimes during the clean-up.

Lathe: This is generally an ordinary six-inch wood turner's lathe on which the zinc shavings are turned by hand.

Pipe Lines for Solution Residues, Laundry, etc., to make up the plant.

Summary of treatment consists in—1. Putting an alkaline solution of caustic wash through the ore to remove the impurities. The strength of this first alkaline solution may be 0.15 per cent cyanide and it may contain up to 8 ounces caustic soda per ton of ore.

2. The Strong Cyanide Solution: This solution varies from 0.3 to 0.5 per cent in cyanide to suit the richness and nature of the material. The strength of this solution should not be less than one-third weight of charge in cyanide.

3. Weak Solution: After the strong solution has been run on we may assume most of the gold is in solution. The object of the third operation is to wash out the dissolved gold, therefore the weak solution containing 0.15 per cent KCN run on the material.

4. In addition to the above treatment the combined treatments can be carried out simultaneously with mercury or amalgamated copper plates and cyanide.

In case of float gold it is sometimes found preferable to float the cyanide and the residue by amalgamation methods.

Class B—Apart from the abnormal consumption of cyanide caused by the materials being acid, a considerable arising from the material, the cause of the abnormal consumption of cyanide is generally prevented by preliminary alkaline treatment, the usual trouble is the presence of certain constituents of copper, iron, etc., and the physical condition in which they exist.

Abnormal consumption of cyanide is usually coupled with poor extraction and the formation of soluble sulphides. The method of determining the cause of the cyanide consumption as usually employed is to examine the solutions: (1) for cyanogen compound; (2) carbonates, oxalates and sulphides; (3) the bases in solution; (4) sulphides in solution; (5) alkalinity.

The general methods of procedure with ores containing cyanides is to subject the material to a water washing which will remove the soluble cyanides, followed by a washing of dilute caustic soda to eliminate basic salts of iron. If this is not efficacious free washing, stimulating Spitzglaster treatment, floating off the bulk of the oxides and hydrates which are present; by this method the light pyritic slimes are also removed which will often considerably reduce the cyanide consumption. Time process, if applied direct, under normal conditions, does not effect satisfactory extraction of the gold, or in the treatment of which the consumption of cyanide is abnormal, or which are exceptionally clayey, etc.

Having broadly classified the ores, etc., the following is an outline of the methods usually adopted for their examination: The material belonging to the first class is crushed to several times the fineness of a part through a 30-mesh sieve—a part through a 40 and a part through a 60. A series of percolation tests are then carried out on the various portions, using cyanide solutions of selected strength. The usual strength for this class of material is from 0.1 to 0.5 per cent and the quantity employed varies from 1.3 to one part for each part of the material to be treated.

The course of the experiments is followed by assaying an average sample of the material as charged into the percolator, and from time to time assaying and testing the solution, the percolator may be by upward and downward flow or entirely by downward flow. For small trials, however, especially on dry materials, it is best that the first percolation be carried out in a wide shallow dish, the remainder of the trial, as by this method channeling is best avoided and a uniform flow of solution over the particles more or less insured.

The normal time of treatment is three to four days, but if the extraction is not fairly complete in that time it is prolonged until such is the case.

Sometimes a better extraction is obtained by a double treatment, i. e., after the treatment has proceeded for a certain length of time, the residues are washed, discharged, recharged to the percolator and treated a second time with fresh solution.

If the material under examination consists of tailings, etc., it is advisable to try in the first instance to obtain satisfactory extraction from it in the condition received.

ed, so as to avoid regrinding; if this is impossible the material should be graded and the various gradings assayed to find the gold, as frequently the gold is found to be locked up in the coarser particles of the residue; on these being finely ground, a further trial often proves satisfactory.

If the material is rich—such as concentrates—the treatment, if by percolation, is often prolonged to a month or so. It is sometimes possible to obtain a better material by the quicker agitation method, but if percolation, even if prolonged, gives a good extraction, it is usually found more economical.

Experiments carried on as above will give sufficient data for a decision as to the best general conditions under which their treatment in practice.

Agitators: These, if used, are worked on day and night shifts and are thus capable of treating three charges per diem of 1.3 ton agitator charges, thus a 50-ton per day agitator plant running on material requiring six hours agitation would comprise seven agitators of about three-ton capacity, treating three charges per day and three 50-ton percolators.

Reservoirs: These should be of sufficient capacity to contain at one time all the solutions, strong and weak, in circulation in the works. As a general rule reservoirs accommodating 1,000 to 1,500 gal. one-third of the cubic contents of the tanks.

Extractors: These are long boxes or troughs containing a prepared fine and so arranged that the gold cyanide solution is forced by baffle boards to pass before it leaves the tank, therefore a 50-ton per day plant will require three extractors of 50-ton capacity and an additional one for contingencies.

They are generally constructed with 10 zinc compartments, each of such dimensions that the amount of solution to be treated may flow up through the zinc easily and steadily.

In practice, each compartment is designed to hold about one cubic inch of zinc for every ton per month treated. Thus a 50-ton plant treating 500 tons per month would have extractors with compartments of such size as to contain about 500 cubic inches of prepared zinc.

This is a vast improvement on the old method of using a six-inch centrifugal wheel well on such a plant.

THE RIFLES HAVE ARRIVED

Soon the Members of the Rossland Company Will be Drilling.

Uniforms Are Now Finished, and Will Follow the Rifles Very Shortly.

The rifles for the Rossland company of the Kootenay rifles have arrived in the city and are now in the office of Captain P. McL. Forin. They are of the latest improved type, and are provided with two sights, one of which is adapted for ranges from 200 to 1,700 yards, and the other from 1,700 to 2,800 yards. The magazine will hold eight or ten cartridges, and the arrangements for recharging the magazine are of the simplest kind. In the stock there is a chamber containing a can of oil, a cleaner and some other things. When the left hand grasps the barrel there is a strip of wood that prevents the heat of the barrel from causing any inconvenience to the riflemen. The rifle is lighter than the Snider rifle, and the sight is a little closer to the eye. Those who have tried them say they shoot very easily and there is but little kick or recoil in them.

The bayonet is a new and somewhat shorter one than the old ones, but is a very formidable weapon. It is attached to the rifle by a small, heavy-duty spring, which when it is slipped on, when it attaches itself with a spring automatically, and can just as easily be removed. The weapon on a whole is a vast improvement on the old ones, and has been heretofore used by the militia of Canada.

The uniforms for the company have been ordered and it is understood that not positively known, that they were already on the way. Just as soon as they arrive here they and the rifles will be served out to the members of the company. This will be in the skating rink as in the articles of incorporation of the Skating Rink company, that is provided for. It is only a question of a few weeks, and before the first of May the Rossland rifles will be parading the streets, headed by their band.

Speaking of the band, it may be necessary for the citizens to provide the necessary uniforms for the band by subscription as the militia department only provides for one band in a battalion, but the people of Rossland are determined to object to furnish them, but will probably take a pride in having the members as handsome as any in the Dominion.

SPREADING ITSELF.

Financial Arrangement for the "Soo" Line—The Corbin Charter.

New York, April 12.—The Canadian Pacific is spreading itself. Yesterday's announcement that it would cut transcontinental time 10 hours, was followed today by the announcement of a big financial arrangement which will enable the Soo line to devote its net earnings to the payment of its interest charges, and at the same time to repay in annual installments the interest which has been advanced by the Canadian Pacific.

The Canadian Pacific has authorized the following expenditures: Improvements of permanent way, \$1,500,000; for additional station yard and terminal facilities at Montreal, Vancouver and other points, \$788,187; for the completion of air brake and automatic coupler equipment, \$305,000; for branch lines to mines in connection with Crown's Net Pass line, \$300,000; rolling stock, \$1,000,000, as the traffic of the company may require.

It all goes to show how well the Canadian Pacific is fortifying itself. The American competitors will at no point get ahead of it very much. It is as strong in Canada as ever, and there is apparently no chance of a charter being granted for the so-called C. C. Corbin line in British Columbia, in which the Great Northern is supposed to be most interested at this time.

Granite and Syenite. Granite and syenite look much alike to a casual observer. They cannot be distinguished by the texture of the surface. The constituents of granite are quartz, orthoclase (potash) feldspar and mica, either black or white. This is simple granite; but in the case of syenite, mica is replaced by minerals, most of which cannot be distinguished without the aid of the microscope. Syenite consists of orthoclase feldspar and hornblende—quartz not an essential constituent. It often occurs in Diorite differs from the above in the character of the feldspar. Its essential constituents are hornblende and plagioclase (soda-lime) feldspar. When quartz occurs it is called quartz diorite, and when mica is present, as well, it is known as quartz-mica-diorite. Magnetite often occurs so abundantly in the fine-grained variety as to give it a black color. The Diorites range in color from light gray to green and black. Typical rocks are rare. They usually graduate from one variety to another, and the name is usually applied in connection with the most prominent constituents. In California the granite of the Sierra Nevada contains the essential constituents of both typical granite and typical diorite; hence the name quartz diorite. Phonolite is usually rather fine-grained, dark and compact, and when struck gives out a ringing sound. It would be impossible to identify the rock without the aid of petrography. Porphyry is a name given to the granitic rocks of the new forming minerals. In the Black Hills' phonolite the principal constituents are nepheline, sanidine, aegirine, noselite and sodalite—minerals familiar only to students of petrology. Porphyry is a name employed to designate a rock having distinct crystals occurring in a ground mass of finer grain, as quartz porphyry and granite porphyry, meaning a granite of fine grain, in which occur large crystals of feldspar. Porphyry does not mean any particular kind of rock, but has reference to the structure only.

A Flexible Currency Based on Gold. New York, April 8.—A declaration for a flexible currency, based on the gold standard, was adopted today by the executive committee of the Indianapolis monetary conference. The meeting of the monetary conference was held at the chamber of commerce of the state of New York, and was called by the chairman, H. H. Hanna of Indianapolis, for the purpose of giving the advice of the committee upon the proposed currency plan to the monetary commission to be presented to the republican caucus committee of the House of Representatives, which meets at Atlantic City on April 17th.

A NATURAL RESULT. Farmer, Wife and Daughter Killed by Dynamite Explosion.

Veederburg, Ind., April 8.—The wife and daughter of John Carrington, a farmer, living about eight miles west of here, were killed by a dynamite explosion. Carrington had prepared about 75 pounds of dynamite under his stove to blow it out when it exploded and blew the house into ruins.

The Fire Hall's Telephone. A new telephone has been installed in the fire hall on Lincoln street, and Chief Guthrie will see to it that alarms are answered from that as well as the regular alarm. The telephone, which is numbered "100," will, he expects, be a useful auxiliary to the fire alarm system.

Spokane Falls & Northern

Nelson & Fort Sheppard R'y

RED MOUNTAIN RAILWAY

The Only Direct Route to Nelson Kaslo, Kootenay Lake and Slocan Points.

Every day in the year between ROSSELAND, ROSSLAND AND NELSON.

EFFECTIVE 12:01 A. M., JAN. 3. LEAVE DAILY. ARRIVE DAILY. 8:20 a. m. Spokane. 6:30 p. m. Nelson. 1:55 p. m. Rossland. 1:00 p. m. Kaslo. 1:55 p. m. Slocan. 1:55 p. m. Slocan. 1:55 p. m. Slocan. No change of cars between Spokane and Rossland.

Tickets on sale all over the world. Close connections at Nelson with steamers for Kaslo and all Kootenay lake points. Passengers for Kettle river, Boundary Camp and Boundary Creek connect at Marcus and Campbell with stage daily.

E. W. RUFF, Agent, Rossland, B. C. SEATTLE & DRYAN, Agents, Trail, B. C. C. G. DIXON, G. F. T. A., Spokane, W. T.

C.P. & N.

THE ONLY LINE EAST VIA SALT LAKE AND DENVER.

SHORTEST AND QUICKEST ROUTE TO—

Coeur d'Alene Mines, Palouse, Lewiston, Walla Walla, Baker City, Portland, San Francisco, Cripple Creek Gold Mines, and all points East and South. Only line East via Salt Lake and Denver. Steamship tickets to Europe and other foreign countries.

Leaves Daily. Spokane Time Schedule. Arrives Daily.

LOCAL MAIL—For Coeur d'Alene, Pomeroy, Walla Walla, Portland, Lewiston, and all points East and South. 7:45 a. m. LOCAL MAIL—From San Francisco, Portland, Walla Walla, Dayton, Pomeroy, Coeur d'Alene, Lewiston, and all points East and South. 7:45 a. m.

FAST MAIL—For Moscow, Pullman, Dayton, Walla Walla, Portland, San Francisco, Baker City and the coast. 3:15 p. m. FAST MAIL—From Baker City, Lewiston, Coeur d'Alene, Moscow, Coeur d'Alene, and the coast. 1:30 p. m.

STEAMSHIP LINES.

San Francisco-Portland Route. STEAMSHIP SAILS FROM AINSWORTH DOCK, Portland, at 8:00 p. m., and from Spear Street Wharf, San Francisco, at 10:00 a. m., every five days.

Portland-Atlantic Line. MONTHLY SAILINGS BETWEEN PORTLAND AND the principal ports of China, Japan and the direction of Dowell, Carill & Co., general agents.

Snake River Route. Steamers between Riparian and Lewiston leave Riparian daily at 3:30 p. m.; returning leave Lewiston daily at 3:30 p. m.

For through tickets and further information apply to any agent S. F. & N. system, or at O. R. & N. Co.'s office, 430 Riverside Ave., Spokane, Wash.

H. M. ADAMS, General Agent, W. H. HURLBURT, Gen. Pass. Agt., Portland, Ore.

ATLANTIC S. S. LINES

FROM ST. JOHN.

Allan Line. Laurentian. April 23
Dominion Line. Vancouver. April 16

FROM NEW YORK.

White Star Line. Teutonic. April 19
Canadian Line. Campania. April 22
American Line. Ethiopia. April 19
Allan Star Line. Mongolian. April 19

FROM BOSTON.

Dominion Line. Dominion. April 19

Passengers arranged to and from all European ports. For rates, tickets and full information call on the public and our trains are operated so as to make close connections with diverging lines at all junction points.

Pullman Palace Sleeping and Chair Cars on through trains. Dining Car service unexcelled. Meals served à la carte.

In order to obtain this first class service, ask the agent to sell you a ticket over the

Kaslo & Slocan Railway

Subject to change without notice. Trains run on Pacific Standard time.

GOING EAST. LEAVE 8:30 a. m. Kaslo. 1:00 p. m. Slocan. 3:00 p. m. Slocan. 5:00 p. m. Slocan. 7:00 p. m. Slocan. 9:00 p. m. Slocan. 11:00 p. m. Slocan. 1:00 a. m. Slocan. 3:00 a. m. Slocan. 5:00 a. m. Slocan. 7:00 a. m. Slocan. 9:00 a. m. Slocan. 11:00 a. m. Slocan. 1:00 p. m. Slocan. 3:00 p. m. Slocan. 5:00 p. m. Slocan. 7:00 p. m. Slocan. 9:00 p. m. Slocan. 11:00 p. m. Slocan. 1:00 a. m. Slocan. 3:00 a. m. Slocan. 5:00 a. m. Slocan. 7:00 a. m. Slocan. 9:00 a. m. Slocan. 11:00 a. m. Slocan. 1:00 p. m. Slocan. 3:00 p. m. Slocan. 5:00 p. m. Slocan. 7:00 p. m. Slocan. 9:00 p. m. Slocan. 11:00 p. m. Slocan. 1:00 a. m. Slocan. 3:00 a. m. Slocan. 5:00 a. m. Slocan. 7:00 a. m. Slocan. 9:00 a. m. Slocan. 11:00 a. m. Slocan. 1:00 p. m. Slocan. 3:00 p. m. Slocan. 5:00 p. m. Slocan. 7:00 p. m. Slocan. 9:00 p. m. Slocan. 11:00 p. m. Slocan. 1:00 a. m. Slocan. 3:00 a. m. Slocan. 5:00 a. m. Slocan. 7:00 a. m. Slocan. 9:00 a. m. Slocan. 11:00 a. m. Slocan. 1:00 p. m. Slocan. 3:00 p. m. Slocan. 5:00 p. m. Slocan. 7:00 p. m. Slocan. 9:00 p. m. Slocan. 11:00 p. m. Slocan. 1:00 a. m. Slocan. 3:00 a. m. Slocan. 5:00 a. m. Slocan. 7:00 a. m. Slocan. 9:00 a.