

Application of belt conveyors and determination of the main parameters of mobile complexes for the transportation of overburden rocks of the Angren coal mine

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ABSTRACT

This article provides information on the use of cyclical-flow technologies at the Angren coal mine of the Republic of Uzbekistan. The method for determining the development time and productivity of the complex during the development of overburden benches in the Angren coal mine is presented. The technological scheme of the development of benches by longitudinal entry of a mobile excavator-crushing complex with a lateral location of the bottoming conveyor and the presence of a mobile inter-step reloader is considered.

Key words: cyclic-flow technology, mobile complexes, belt conveyor, excavator-crushing complex, complex productivity.

1. INTRODUCTION

The Angren coal mine is located on the territory of the Tashkent region of the Republic of Uzbekistan and is located in the valley of the Akhangaran river between the Chatkal and Kuramin ranges. The field stretches from southwest to northeast for 9 km and from southeast to northwest for 15 km. The total area of the deposit is about 70 km².

The Angren open-cast mine is a unique kaolin - coal deposit. Overburden rocks (associated minerals) lie at the top of the coal seam: secondary gray kaolins, secondary variegated kaolins, calcareous sandstones, siltstones, quartz-mica sandstones, dolomitized and pure limestones, marls, loams, pebbles, loess. Mining and geologic conditions of the bedding of the Angren coal mine field predetermined the use of the transport system of development in the section with the movement of overburden into external and internal dumps by means of railway, road and conveyor transport.

The heights of the mining and overburden benches were up to 15 m. The width of the working platforms varied within 35-100 m. As a result of the impact of landslide processes in the southwestern, western and northwestern sections, the width of the working platforms of coal and kaolin benches

was from 50 to 60 meters. The direction of the work front is from south to west. Mining operations at the "Upper" complex are carried out selectively using ЭКТ-4y excavators with loading onto road transport. At the same time, the ash content of coal reaches 30-35%. Mining operations at the Angren coal mine open being carried out at the Upper and Powerful complexes 61 % of the mined coal falls on the "Powerful" complex, 39% - on the "Upper" complex. At the "Powerful" complex, mining benches are worked out by excavators with loading on conveyor and road transport, where excavators ЭКТ-5А and ЭКТ-4y operate [1,2,3].

In accordance with the established assignment for the Angren coal mine Open-Pit Branch for coal production in the amount of 3,200.0 thousand tons, taking into account the implementation of the Government Program for the provision of solid fuel to the regions of the Republic of Uzbekistan, the following main types of stripping operations are envisaged [3]:

stripping works, total - 29500 thousand m³,
including, by type of transport:
railway - 9,000 thousand m³.
automobile transportation - 12,000 thousand m³.
With out transportation - 2,500 thousand m³.
Cyclic-flow technology - 6,000 thousand m³.

In March of the 2015 year, it was received and installed. The excavator type of ЭКТ-15М was put into operation. In addition, in 2014 year 4 conveyors of the central heating center of the coal direction and the loading and sorting point

(LSP) "Dzhigiristan" were put into operation.

At the Angren coal mine, a technological scheme for organizing mining operations is used in the development of overburden [2,4,5]: with the installation of a mobile conveyor-reloader or inter-step conveyor-reloader between the mobile crusher and the face conveyor (Fig. 1). The rock is loaded by the excavator into the crushers through the feed hopper. Rock from the crushing plant is transferred to a mobile transfer conveyor and then to the face conveyor. The use of a mobile conveyor-reloader allows to reduce the frequent movement of the face conveyor and place them at a great distance from the face;

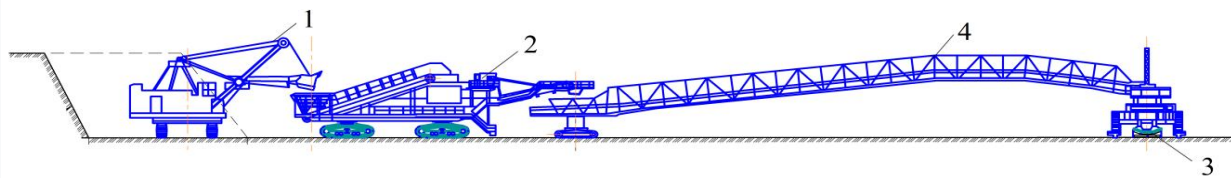


Figure 1: Schemes for the development of rocks using conveyor transport 1 - excavator; 2 - mobile crushing plant; 3 - face belt conveyor; 4 - mobile conveyor-reloader;

2. METHODOLOGY

Cyclic flow technology at the overburden complex consists of the following technical links [2,8,12]:

- excavator (ЭКТ-15) - 3 units;
- a mobile crushing plant receives rock mass from an ЭКТ-15 excavator with its subsequent loading onto a downhole inter-step reloader with with performance of $Q = 4000$ t/h - 3 units;
- downhole bridge-type crusher, used as a connecting link between the mobile crusher and the face conveyor and transferring the crushed rock mass from the mobile crusher to the face belt conveyor with a working bench height of 15 m and with performance of $Q = 4000$ t/h - 3 units;
- belt conveyor used to transport rock mass from a mobile crusher to a spreader with a total number of overburden belt conveyors 14;
- a spreader designed for dumping overburden into an internal dump with an unloading boom length of 60 m, a total length of 110 m and with performance of $Q = 12100$ t/h - 1 unit.

The sequence of work in the central heating center is carried out as follows: the ЭКТ-15 excavator loads the overburden into the bunker of the crushing plant, then the overburden from the bunker enters the plate conveyor, from there, through the hopper, it enters a two-roll crusher, which passes through itself the transported material with a size of 1100mm at the exit up to 300mm. The overburden goes through the crusher's outlet chute to belt conveyor, which transports it to a mobile transfer loader.

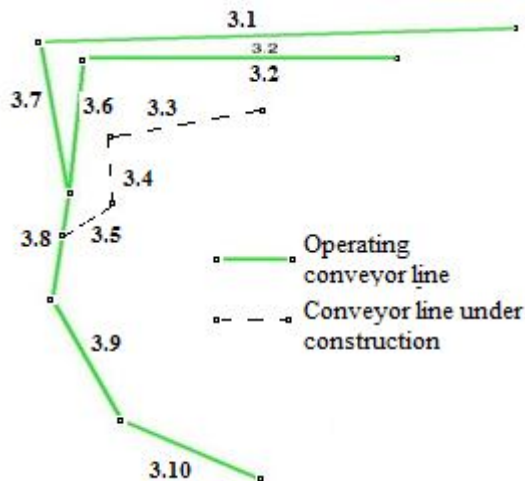


Figure 2: Layout of overburden conveyors

Further transportation of the rock mass is carried out by a main conveyor with a capacity of 12100 t/h, followed by reloading onto a dump conveyor, from where the rock mass moves to a spreader, which forms internal dumps. Details of the technical parameters of the conveyors are given in table. 1. The layout of the constructed 1 and 2 lines and the 3 line under construction is shown in Fig. 1.

Overburden transportation:

CFT of the 1-tier.

A face conveyor № 3.1 pours overburden onto the transfer conveyor №3.7, which in turn transports the soil to the transfer onto the main conveyor № 3.8 (Fig. 2).

With the advancement of the work front, the transfer conveyor №3.7 is built uppreserving its initial bearing azimuth.

CFT 2-tier.

A face conveyor № 3.2 pours overburden onto the transfer conveyor № 3.6 and then onto the main conveyor № 3.8. To ensure the advancement of the work front, the intermediate conveyor № 3.6 will be built up

CFT 3-tier.

A face conveyor № 3.3 horizon pours overburden onto the transfer conveyor № 3.4, which in turn transports the soil to the transfer conveyor № 3.5. Further, on the transfer conveyor, the soil is delivered to the dump conveyor № 3.8.

Ensuring the advancement of the work front is achieved by building up the transfer conveyor № 3.4.

The belt conveyors that are used at the Angren coal mine have structures of a stationary type, a semi-stationary type and a mobile structure with rails [5].

The head station and tail station of the conveyor, which has a semi-stationary type structure, are located on steel skis, the middle section is located on short steel sleepers, and hooks are laid on the side of the middle section for fastening the cable.

The head station and tail station of the conveyor, which has a mobile structure with rails, are located on steel skis, the middle section is located on long steel sleepers on which the rails are laid, and a loading carriage with a carriage with a cable drum and a dump carriage are equipped, on board the middle section are laid hooks for fastening the cable. A characteristic feature of the use of the central heating center is the flow of the entire technological chain, the stoppage or failure of the operation of individual links

or devices, leads to the stoppage of all lines of the complex. To exclude such situations, the strictest control over the implementation of technical operation standards is

introduced. Another way to anticipate emergencies is early diagnosis and anticipated malfunctions and repairs.

Table 1: The actual length of the conveyor lines of the overburden CFT of the Angren coal mine branch

№	Conveyor	Actual length (m)	Actual tilt angle (degree)	Installation horizon of the drive station	Take-up station installation horizon	Angle of rotation (degree)
1	3.1	960	0,20	+ 955m	+ 950 m	79
2	3.2	520	1	+ 926 m	+ 915 m	110
3	3.6	354	1	+ 931 m	+ 926 m	173
4	3.7	855	1	+ 933 m	+ 955 m	169
5	3.8	806,5	0,74	+ 941 m	+ 931 m	-
6	3.9	631	0,47	+ 936 m	+ 941 m	159
7	3.10	350	0	+ 936 m	+ 936 m	163
Total		4476,5				

The control system of the CFT complex operates identically to similar systems and includes an automated workstation for the user. All links of the complex are connected into a single network, the process of which is displayed on the computer monitor. From the central control center, the user visually monitors the work of the central heating center in real time.

The problems of improving the cyclical-flow technology in open-pit mining are due to the mining conditions of open pits. These conditions determine the need to develop new technological and technical solutions for the adaptation of the CFT, taking into account the specific mining and geological conditions of the open pit.

World practice in open pit mining shows that in large open-pit mines the use of various types of CFT is becoming more and more widespread. Taking into account

the extensive experience of using this technology in various deposits, the economy in the transportation of rock mass and energy consumption, saving human resources, it is not difficult to foresee that the future development of open pit mining is behind this technology.

Let us consider the technological scheme for the development of rocks by the end-face arrangement of mobile crushing, reloading and conveyor complexes (MCRCC) using a single-bucket excavator with a block length $L = 500$ m and the duration of the complex operation for 1 year (Fig. 3) [3,11].

The minimum width of the working platform should be equal to one entry of the excavator, taking into account the overall dimensions of the crushing plant.

Table 2:List of technical parameters of stripping belt conveyors of the central heating center of the Angren coal mine

№№ п/п	№ conveyor	Conveyor name	Belt width, mm	Productivity, t/h	Belt speed, m/s	Conveyor length, m	Conveyor tilt angle, degrees	Number of drives, pcs.	Conveyor classification
1	2	3	5	6	7	8	9	10	11
1	3.1	Face conveyor	1200	4000	5,0	1800	0°	4	mobile with rails
2	3.2	Faceconveyor	1200	4000	5,0	1700	1°	4	mobile with rails
3	3.6	Transfer conveyor	1200	4000	5,0	460	3°	2	Semistationary
4	3.7	Transfer conveyor	1200	4000	5,0	500	0°	1	Semistationary
5	3.8	Main conveyor	2000	12100	5,6	500	1°	3	Semistationary
6	3.9	Dump conveyor№1	2000	12100	5,6	700	1°	4	mobile with rails
7	3.10	Dump conveyor№2	2000	12100	5,6	650	1°	4	mobile with rails
		Total:				6310		22	

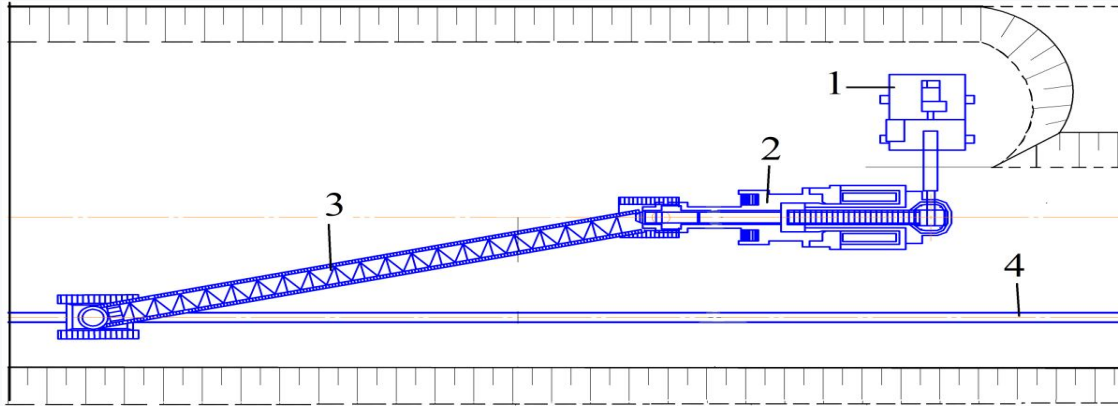


Figure 3:Technological scheme for the development of rocks with the end location of MCRCC with an increased width of the working platform: 1 - excavator; 2 - mobile crushing plant; 3 - mobile conveyor loader; 4 - face conveyor

3. METHODOLOGY FOR DETERMINING OPERATION TIME AND PRODUCTIVITY OF THE MCRCC COMPLEX

The width of the strip at the same time taken out is determined by the formula [6,11]:

$$B_n = R_d + R_u + l_{uc} + l_{lc} - C_1, \text{m}, \quad (1)$$

R_d - excavator digging radius at the standing level, m;

R_u - excavator unloading radius, m;

l_{uc} - departure of the discharge cantilever conveyor of the MCP, m;

l_{lc} - effective length of the loading conveyor, m;

C_1 - the minimum safe distance from the edge of the camber to the conveyor line, m.

Let us determine the annual productivity of the MCRCC complex according to this technological scheme (Fig.3).

The total annual productivity of the MCRCC system depends on mining-geological, mining-technical, technological and organizational factors and is determined taking into account the advancement of the mining front [6,12,13,14]:

$$Q_c = U_f L_b h, \text{m}^3/\text{year}, \quad (2)$$

U_f - length of advance of the front of mining operations, m;

L_b - block length, m ($L_b = 500$ m); h - bench height, m.

In the technological scheme of the development of rocks by the end position of the MCP with the use of a single-bucket excavator and conveyor transport, it takes a long time to move the face conveyor along the front. In this regard, when determining the annual productivity of the complex, the time of moving the face conveyor together with the development time of the entire length of the block is taken into account.

The working time of the complex during the development of one block (cycle) is

$$T_{CYCLE}^1 = \frac{V_{BLOK}}{Q_{oper} K_P} = \frac{B_W L_B H_L}{Q_{O.P.} K_P}, \quad (3)$$

where Q_{oper}^h - is the operational productivity of the complex, m^3/h ; K_P - coefficient of productivity reduction, taking into account the shutdown of the complex, technological and organizational factors ($K_P = 0.85-0.95$); V_{BLOK} - the volume of the processed block, m^3 ; B_W - excavation trench width, m; L_B - block length, m; H_L - ledge height, m.

The time of movement of the downhole conveyor is determined by the formula [8]

$$T_{D.C.M.} = t_p + \frac{B_W L_d K_i}{K_S^t P_T} + t_f, \quad (4)$$

where t_p - is the time for preliminary and auxiliary operations when moving the conveyor stand ($t_p = 3$ h according to [8]);

L_d - the length of the stav downhole conveyor, m;

K_i - coefficient taking into account the influence of production conditions on the implementation of the movement process ($K_i = 1,2$ according to [8]);

K_S^t - coefficient of use of the tournode doser during the shift ($K_S^t = 0.56$ according to [3]);

t_f - time for final operations when moving the conveyor bed ($t_f = 4$ hours according to [8]); $P_T P_T$ - technical productivity of the tournode doser

$$P_T = v_T + B_T, \text{m}^2/\text{h} \quad (5)$$

where v_T - operating speed of a tournode doser, m/h;

B_T - step width of the stroke of the tournode doser, m

The annual operating time of the complex, taking into account the working time of the excavator, is determined by the formula:

$$T_A = T_{SH} n_{SH} N_W K_U, \quad (6)$$

where T_{SH} - the duration of the shift, h;

n_{SH} - the number of shifts per day;

N_w - the number of working days of the excavator per year;
 K_U - utilization factor of the complex change time (is $K_U = 0,73-0,85$).

The number of cycles performed in the developed blocks of the complex per year is determined by the formula:

$$N_{CYCLE}^1 = \frac{T_A}{T_{CYCLE}^1 + T_{m.d.k}}$$

The annual productivity of the complex is determined by the formula:

$$Q_{COM}^A = V_{BLOK} \cdot N_{CYCLE}^1 = B_W L_B H_L \cdot N_{CYCLE}^1, m^3/year$$

4. RESULTS

In the table 3 shows the results of calculations of the annual productivity of the MCRCC complex at different block lengths according to the 1st option of the technological scheme.

According to the results of calculations, according to table. 3, the dependence of the annual productivity of the MCRCC complex on the block length was obtained, which is shown in Fig. 4.

Table 3: Calculation results of the MCRCC complex during the development of benches according to the technological scheme for the development of rocks with the end face of the equipment using a bucket excavator and conveyor transport

$Q_{COM}^A, m^3/y$	$Q_{oper}^h, m^3/h$	V_{BLOK}, m^3	T_{CYCLE}, h	L_{BLOK}, m	$T_{D.C.M.}, h$	T_A, h	N_{CYCLE}^1	$\Delta t_{п.з.к.}, \%$
4767252,6	1478	53400	40,14	200	19,71	5344	89,3	0,099
4960630,4	1478	80100	60,21	300	26,07	5344	61,9	0,087
5063324,2	1478	106800	80,28	400	32,43	5344	47,4	0,081
5127007,1	1478	133500	100,4	500	38,79	5344	38,4	0,078
5170359,8	1478	160200	120,4	600	45,14	5344	32,3	0,075
5201777,6	1478	186900	140,5	700	51,5	5344	27,8	0,074
5225592,7	1478	213600	160,6	800	57,86	5344	24,5	0,072
5244266,9	1478	240300	180,6	900	64,21	5344	21,8	0,071
5259302,5	1478	267000	200,7	1000	70,57	5344	19,7	0,071
5271668,7	1478	293700	220,8	1100	76,93	5344	17,9	0,070
5282018,4	1478	320400	240,8	1200	83,29	5344	16,5	0,069
5290807,6	1478	347100	260,9	1300	89,64	5344	15,2	0,069
5298364,6	1478	373800	281	1400	96	5344	14,2	0,069
5304931,4	1478	400500	301,1	1500	102,4	5344	13,2	0,068
5310690,7	1478	427200	321,1	1600	108,7	5344	12,4	0,068
5315782,9	1478	453900	341,2	1700	115,1	5344	11,7	0,068
5320317,5	1478	480600	361,3	1800	121,4	5344	11,1	0,067

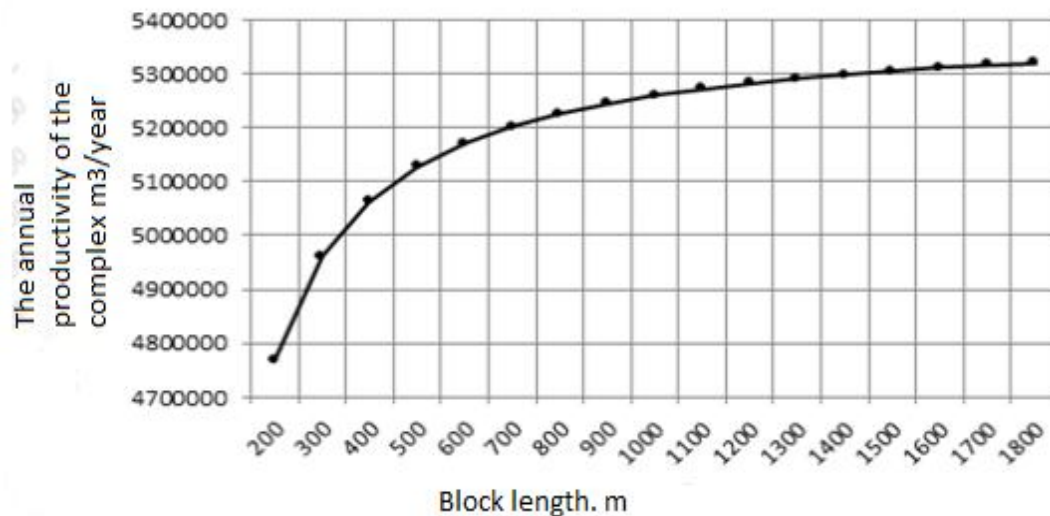


Figure4. Dependence of the annual productivity of the MCRCC complex on the length of the block and a comparison of three variants of technological schemes

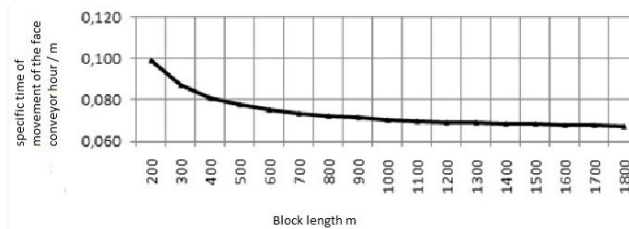


Figure 5. Dependence of the change in the specific time of movement of the face conveyor on the length of the block.

5. CONCLUSION

The results obtained show that the productivity of the MCRCC complex increases with increasing block length. This is due to a decrease in the number of cycles during the period of operation throughout the year.

Studies have also found that the productivity of the MCRCC complex increases with a decrease in the specific time for moving the face conveyor per working cycle (Fig. 5).

The novelty of this technological scheme is the movement of the complex following the movement of the excavator in the face and thereby ensuring the flexibility and mobility of the entire mining and transport system. In combination with continuously operating conveyors, the mobile crushing complex makes it possible to abandon the alternative excavator-motor transport scheme and the fleet of heavy-duty dump trucks. In addition to reducing mining costs, the development technology based on the new fully mobile crushing system also provides significant reductions in exhaust gas emissions into the atmosphere.

Thus, the recommended process flow diagram of the overburden mining center with the use of mobile crushing-transfer-conveyor complexes ensures the dynamic development of mining operations along the front and depth with high technical and economic indicators of open pit mining.

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