Analysis of the Karst Development Law Based on Multiple Exploration Technologies of Cambrian Limestone

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Research

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Abstract

Coal mine floor limestone aquifers are a major source of water inrush from the coal seam floor and a serious threat to the safety of coal mining. In order to reduce and avoid the occurrence of water inrush within the coal mine, we use multiple detection techniques, which are geophysical exploration technology, drilling technology, water inrush accidents and tracer test, to develop a multi-faceted exploration of karst development and analyze its development characteristics in the Chaochuan mine No. 1 well of Pingdingshan Coal Co.Ltd, Henan Province, China. The results show that, the Cambrian limestone (CL) karst water is poor; there is a certain hydraulic connection. Near faults F5, F1, and SF28, the area is less water-rich area, the water is weak, and the deep karst water forms a closed area; 61.54 % of shallow water inrush accidents in the Taiyuan limestone and CL karsts were caused by large tectonic and nearby shallow faults. The karst vertical zonation is shallow; the shallow water level decreased more in the West Wing of the No. 1 well than in the East Wing at elevations above 140 m and below -150 m. The F125 level decline is greater than that of the east, and the west fault is more than 170 m due to the hydraulic connection intrusive barrier wings of the karst water. the East Wing below -150 m F125 fault was weak and uneven on both sides of the hydraulic connection.

1. Introduction

Water inrush from the coal seam floor has always threatened safe production in mines. The water inrush is not only sudden, but also has a strong impact. In a short time, it can flood wells and bring huge economic losses [1–3]. Limestone is often hidden at the base of the coal seam. Due to its karst development, strongly water-rich with good connectivity, limestone is the main water source for the coal seam floor under mining conditions, seriously threatening safe mining of coal [4–6].

The karst development of the coal seam floor is affected by many factors and has very complex features such as hydraulic connections and water inhomogeneity. How to accurately depict the characteristics of karst development and the water-rich law is a major issue in the study of mine hydrogeology [7, 8]. Wang [9, 10] have studied the water rich characteristics of the Cambrian limestone in the No. 2 well of the Pingdingshan coal field and formulated the related countermeasures for water control by using geological drilling, field connectivity tests, water drainage drilling, water temperature field monitoring, and transient electromagnetic exploration. Dai [11] studied the influence of geological structures on karst development in the Hancheng mining area and concluded that: (1) the structural fissures are dominant in karst development, and (2) the vertical karst development shows a zonal distribution where upper and lower aquifer connectivity is strong. Hao [12] analyzed the karst development characteristics and influencing factors of the Ordovician limestone in the Gujiao mining area by using the hydrogeological drilling data and their core characteristics, and concluded that the karst development is mainly controlled by the lithologic rhythm combination, resulting in karst phenomenon between the strong and weak phases, where the ground structure raises the stratum and controls the degree of karst development. Hu [13] applied chemical tracer tests to explore the characteristics of deep karst development in the Qiuji mine and concluded that the karst development in the mine field is not uniform. Most of the areas are strong karst regions, and the karst development in a few regions is mainly characterized by fissure and weak karst development. The above research on the characteristics of karst development and the achievements made in the typical coal mine have laid a scientific foundation for the identification of water inrush sources and the probability of water in-rush occurring, and previous studies have also helped determine the direction for prevention and control of the coal mine water damage, which has important theoretical and practical significance.

This paper takes the No. 1 well in the Chaochuan mine of Pingdingshan Coal Co.Ltd, Henan Province, China as an example (abbreviated as the No. 1 well). Based on the analysis of the hydrogeological conditions of the mine, geophysical exploration, hydrogeological drilling, water in-rush feature analysis, underground water discharge
engineering, dynamic changes of groundwater level, and tracer tests were adopted. A comprehensive analysis of the Cambrian limestone (abbreviated as CL) karst development characteristics of the No. 1 well coal seam floor was performed, and then the water strength in order to provide a beneficial reference for the prevention and control of mine water hazards in the future and reduce the probability of water inrush accident in karst aquifer of coal seam floor.

2. Geological Overview

Chaochuan mine is located about 15 km south of Chenzhou City in Henan Province. The mine measures about 10 km from east to west and 4 km from south to north, and the mining area is about 21 km². The No. 1 well is located in the middle of the Chaochuan mine, with fault F₅ in the north, coal outcrop in the south, fault F₁₀₄ in the southwest, and fault F₁ in the northeast. It accounts for 34.4% of the total area of the Chaochuan mine (Fig. 1).

The main faults of the No. 1 well are located on the northern, western, and southern boundaries of the well field, and a series of northwest (NW) and near east-west (E-W) faults play an important role in the burial and distribution of the CL aquifer on the coal seam floor of the minefield and the discharge of groundwater recharge and runoff (Fig. 2).

The F₁ normal fault extends 4,300 m in the No. 1 well and with a drop of 60 m to 130 m. Because the lower CL aquifer of the F₁ fault is connected with the No. 2 – 1 coal seam in the upper plate, the CL is buried deeper in the lower part, and the karst development is weak. The F₁ fault is a lateral non-conducting fault, or a weak water conducting fault, in the No. 1 well. However, in the fracture zone of the upper wall, there may be a water-rich strip along the strike of the fault.

The F₅ normal fault in the No. 1 well intersects the F₄ faults, with an extension of 3,150 m and a drop of 26 m to 118 m. The fault causes the CL in the No. 1 well to stagger along the deep part. It is a non-conducting or weakly conducting boundary for the No. 1 well.

The SF₂₈ drop is at 40 m to 130 m, and the extension length is 3,600 m. SF₂₈ causes the CL in the south to connect with the coal seams or Carboniferous limestone in the north. Thus, it loses the hydraulic connection with the CL in the north, and the lateral water conductivity of the fault is weak. The runoff of the CL groundwater is blocked northward or northeast (NE), resulting in the formation of a water-rich zone on the southern plane of the fault.

The F₁₂₅ normal fault extends 850 m in the No. 1 well with a drop of 20 m to 70 m. The underground water of the CL, found in the two disks on the East and west sides of the fault, is connected. Therefore, the shallow part belongs to the lateral water diversion fault, while the groundwater hydraulic relations between the two pieces of CL in the deep fault are weak.

The F₁₀₄ normal fault is a fault exposed underground, extending 1,600 m with a drop of 50 m.

The No. 1 well coal-bearing stratum is shown in Fig. 3. The leading role of No. 1 well water-filling is the coal seam floor, which consists of the Carboniferous Taiyuan formation thin limestone and Cambrian thick limestone aquifers. The L₁ +₂ thin limestone in the Taiyuan formation is generally developed, with an average thickness of about 3.83 m, and it is about 8 m from the bottom of the No.2 – 1 coal seam. The average thickness of the Cambrian thick limestone is 146.24 m, with an elevation of -100 to -600 m (Fig. 2). The aquifer is about 50 m from the bottom of the No. 2 – 1 coal seam. Due to the well-developed geological faults in the No. 1 well, the karst fissures in the CL aquifer are also well developed as a result of the influence of faults and the floor damage caused by mining. The groundwater is not only the main supply source of the limestone aquifers in the upper Taiyuan group but also the main water filling source of the No. 2 – 1 coal seam mining.
3. Transient Electromagnetic Prospecting

In the main body of the paper, three different levels of headings (for sections, subsections, and subsubsections) may be used. The typesetting style for these headings is presented in the next section.

In order to discover the water-rich anomalies of the CL aquifer in the No. 2 − 1 coal seam, ground transient electromagnetic [14, 15] exploration was carried out within the No. 1 well range. The following exploration range was used: From the F5 reverse fault southward to -250 m water storehouse, and west from the 13th exploration line eastward to the 5th exploration line. The total area was 3.1 km2, with a survey line spacing of 40 m * 40 m and a total of 3,184 physical exploration points. The detection depth was set at 80 m in the CL aquifer. According to the transient electromagnetic apparent resistivity, the aquifer was usually divided into 2 levels, rich in water and weak in water [9, 10]. According to the exploration situation in this mine, in the top range of the CL (depth 0 ~ 40 m), an apparent resistivity of less than 85 Ω•m was defined as a strong water-rich zone; and in the middle of the CL depth (depth 40 ~ 80 m), apparent resistivity less than 90 Ω•m was defined as a strong water-rich area. The survey results are shown in Fig. 4.

From Fig. 4, the depth of the 0 to 40 m CL has been delineated in 14 low resistivity anomalies, and the depth of the 40 to 80 m CL has been delineated in 16 low resistivity zones. The proportions of the areas occupied by the survey were 17.94% and 18.42%, respectively. In addition, the distribution of the low resistivity anomaly areas was sporadic and beaded, and the range of a single low resistivity anomaly was small, which indicated that the CL aquifer in the No. 1 well range had poor water-richness as a whole. In the 0 ~ 40 m and 40 ~ 80 m depths from the top interface of the CL, there were overlapping regions in the low resistivity anomaly area, which indicated that there was a certain hydraulic connection between the groundwater in the top and the middle sections of the CL aquifer.

In addition, near F5, F1, and SF28, the low-resistance anomaly area was sporadic, indicating that these faults were weakly conducting water, which was consistent with the actual situation revealed during the construction of the mine roadway. In addition, due to the weak water conductivity of F5, F1, and SF28, the hydraulic connection between the deep and shallow parts of the CL was weak on both sides of the fault. A relatively closed hydrogeological unit of the No. 1 well CL aquifer was formed.

4. Hydrogeological Exploration

During the No. 1 well exploration, 10 geological holes were drilled from the target layer to the CL (Table 1). The drilling depth was between 16.64 and 652.33 m, and the mean value was 342 m. The final borehole elevation was within the range of 250.94~ -415.21 m with a mean value of -106.17 m. The water leakage in all 10 boreholes indicates that the karst fissures in the shallow CL aquifer were generally developed and rich in water.

In the No. 1 well mining process, in order to monitor the groundwater level of the CL aquifer, eight hydrogeological boreholes were constructed at different stages (Table 2). The drilling depth was between 420.20 m and 874.60 m, and the mean value was 555.30 m. The final hole elevation in the borehole ranged from - 242.00 m to -693.00 m with an average of -355.69 m. There is no leakage phenomenon in the 8 boreholes, which indicated that the CL karst in the deep part of the area was weak and not strongly water-rich.

Comparing the boreholes in Tables 1 and 2, we see that the borehole depth of the former was much smaller than that of the latter, and the elevation of the final borehole was much higher than that of the latter. For the exposed CL thickness, the former was 44.51 m, and the latter was 40.84 m. The difference between the two was not large, but the leakage situation is very different. This is due to the large amount of groundwater being discharged during the
excavation process of the mine, resulting in a drastic decline in the groundwater level of the CL aquifers, which made it impossible to see groundwater in the hydrogeological boreholes of later construction.

Table 1

| Drilling number | Leakage point position | Water leakage |
|-----------------|------------------------|---------------|
|                 | Depth(m) | Elevation(m) |                             |
| 1-101           | 239.29   | -67.38       | The exposed thickness of the CL is 56.65 m, and drilling is stopped after leakage |
| 10–97           | 267.25   | -48.44       | The exposed thickness of the CL is 104.37 m, and drilling is stopped after leakage |
| 12–21           | 473.68   | -245.66      | The exposed thickness of the CL is 38.98 m, and drilling is stopped after leakage |
| 12–96           | 370.00   | -156.88      | The exposed thickness of the CL is 44.64 m, and drilling is stopped after leakage |
| 13–93           | 652.33   | -415.21      | Revealing the thickness of the CL is the discovery of water leakage, and then terminates drilling |
| 14–91           | 313.00   | -89.94       | The exposed thickness of the CL is 34.40 m, and drilling is stopped after leakage |
| 14–92           | 444.00   | -228.93      | The exposed thickness of the CL is 12.89 m, and drilling is stopped after leakage |
| 20–23           | 562.88   | -257.07      | The exposed thickness of the CL is 82.35 m, and drilling is stopped after leakage |
| 21–88           | 80.95    | 196.89       | The exposed thickness of the CL is 55.35 m, and drilling is stopped after leakage |
| 27–21           | 16.64    | 250.94       | The exposed thickness of the CL is 15.45 m, and drilling is stopped after leakage |
| Mean            | 342.00   | -106.17      | The average exposed thickness of the CL is 44.51 m |
Table 2
Hydrogeological borehole of the CL

| Drilling number | depth (m) | elevation (m) | Water leakage |
|-----------------|-----------|---------------|---------------|
| Water 1         | 440.00    | -262.00       | The revealed thickness of the CL is 61.50 m, and there is no water leakage phenomenon |
| Water 2         | 420.20    | -242.00       | The revealed thickness of the CL is 3.70 m, with rock breaking, there is no leakage phenomenon |
| Water 3         | 489.00    | -314.00       | The revealed thickness of the CL is 13.00 m, there is no water leakage phenomenon |
| Water 6         | 495.20    | -289.00       | The revealed thickness of the CL is 14.60 m, there is no water leakage phenomenon |
| Water 7         | 715.20    | -468.00       | The revealed thickness of the CL is 7.20 m, there is no water leakage phenomenon |
| Water 8         | 424.60    | -250.00       | The revealed thickness of the CL is 85.20 m, rock breaking, there is no leakage phenomenon. |
| Water 9         | 874.60    | -693.00       | The revealed thickness of the CL is 67.50 m, with rock breaking, there is no leakage phenomenon. |
| 4–99            | 583.62    | -327.50       | The revealed thickness of the CL is 74 m, karst is not developed |
| Mean            | 555.30    | -355.69       | The average revealed thickness of the CL is 40.84 m |

5. Water Inrush Feature Analysis

Since 1973, there have been 22 water inrush accidents in the No. 1 well (Fig. 5). Eighteen of them were water inrush from the limestone aquifer of the bottom plate, accounting for 81.82% of total occurrences. There were 13 times where water inrush exceeded 10 m³/h, and 11 of them where the water inrush originated from the CL aquifer, accounting for 84.62%. The maximum two water inrush occurrences were 1,440 m³/h and 1,996 m³/h, both from the CL aquifer. These results indicated that the most serious threat to No. 1 well mining is the bottom CL aquifer.

For the 13 inrush water incidents with No. 1 well water volume exceeding 10 m³/h, eight were caused by geological structure (fault and fold), accounting for 61.54%. For the eight events, seven were water in-rush from the fault and one from the syncline. Figure 5 shows that most of the water in-rush points associated with the fault were located near the fault facing E-W direction, indicating that the structural fracture zone was a zone of karst fissure, which not only controlled the enrichment of the groundwater but also affected the direction of the runoff. Therefore, it was the most prone position for water inrush in the bottom limestone aquifer.

Secondly, as shown in Fig. 5, there were 18 cases for the water inrush of the limestone aquifer of the No. 2 – 1 coal seam above – 200 m, accounting for 81.82% of the total water inrush accidents, indicating that the shallow limestone aquifer had more water inrush than the deep part. The incidents were also more concentrated in the vicinity of small faults, which was mainly due to the fact that shallow small faults were more developed than deep ones. In general, the large faults mainly caused the breakage of the regional rock formations, while the minor faults made the surrounding rock mass more broken, resulting in the development of karst fractures.

In addition, the fractured rock layer had a large CO₂ content, making the erosion ability stronger and the groundwater circulation in the karst fracture faster, which led to stronger CO₂ erosion ability [16]. With the increase of depth, the CO₂
recharge in the water was not sufficient, and the groundwater circulation slowed down, which caused the erosion ability of CO$_2$ to decrease, so that the karst development in the mine field eventually led to zoning in the vertical direction. That is, the shallow karst development was stronger than the deep part [17, 18].

Based on the analysis of the water inrush frequency, the shallow karst fracture was more developed than the deep part, which was consistent with the third part.

6. Downhole Drainage And Decreasing Pressure Drilling

In order to reduce the underground water level of the limestone aquifer in the Cambrian, which will reduce the threat of water inrush from the floor of the coal seam mining, the No. 1 well-constructed special discharge roadways (Fig. 6) according to the mining progress, and built drainage and decreasing pressure drilling in the roadway (Table 3). According to Table 3, the benefits could be seen from four indices, such as the number of water inrush holes, minimum water inflow, maximum water inflow, and average water inflow. Water inflow must occur in the borehole of the roadway with an elevation of more than $-250$ m, The minimum, maximum, and average water inflows range from $0.01 \sim 16.00$ m$^3$/h, $150.00 \sim 232$ m$^3$/h, and $84.28 \sim 90.00$ m$^3$/h, respectively, The proportions of water in-rush drilling, minimum water inflow, maximum water inflow, and average water inflow in the roadway with elevations below $-300$ m range from $40.74 \sim 100\%$, $0.00 \sim 1.00$ m$^3$/h, $3.00 \sim 90.00$ m$^3$/h, and $2.00 \sim 17.61$ m$^3$/h, respectively. Obviously, the former four indicators were far better than the latter, reflecting that the karst development of the limestone aquifers of the deep Cambrian in the deep No. 1 well was weaker than that in the shallow, which was completely consistent with the conclusions from the previous analysis in Parts 3 and 4.

| Roadway     | Drill hole | Water inflow (m$^3$/h) |
|-------------|------------|------------------------|
|             | Total      | Number of water inrush | Proportion (%) | Minimum  | Maximum  | Mean     |
| -150 m West Wing | 9          | 9                      | 100           | 16.00    | 150.00   | 90.00    |
| -250 m West Wing | 23         | 23                     | 100           | 0.01     | 180.00   | 85.54    |
| -250 m East Wing | 18         | 18                     | 100           | 0.50     | 232.00   | 84.28    |
| -300 m West Wing | 27         | 11                     | 40.74         | 0.00     | 85.00    | 14.96    |
| -350 m East Wing | 5          | 5                      | 100           | 1.00     | 3.00     | 2.00     |
| -390 m West Wing | 37         | 26                     | 70.27         | 0.00     | 90.00    | 17.61    |

7. Analysis Of Time And Space Changes In The Water Level

The location of the groundwater level observation hole for the CL and its water level dynamic changes are shown in Fig. 7.

From July 2000 to July 2006, the drainage of the No. 1 well mainly occurred in the $-150$ m West Wing and the $-250$ m West Wing discharge roadways, and the discharge volume was stable at $800$ m$^3$/h. The 4–92 observation hole, which was close to the two discharge roadways, changed from $+6.0$ m to $-210.0$ m, and continued to decrease by $216$ m. From July 2006 to March 2012, the main drainage was in the $-250$ m West Wing discharge roadway, the drainage...
water volume was 1373.5 m$^3$/h, and the water level of the 4–92 observation hole was basically stable at -220~225 m. With the implementation of the hydrophobic scheme, the water in the East Wing −106 m water inrush point disappeared, indicating that the groundwater of the CL at the shallow elevation of -150 m and F$_{125}$ fault was closely related to underground water.

From July 2006 to March 2012, while the −250 m West Wing drainage roadway was draining, drainage also occurred in the −250 m East Wing drainage roadway, with a drainage volume of 2,580.0 m$^3$/h. The water level of the 4–99 observation hole on the east side of the F$_{125}$ fault decreased from −11.0 m to -42.2 m, or only 31.2 m. The water level of the S1 observation hole decreased from −29.2 m to -60.5 m, which only decreased by 31.3 m, reflecting the poor water conductivity of the F$_{125}$ fault as a whole. These results showed that the groundwater hydraulic connection of the CL on both sides of the fault was not very close.

From April 2012 to September 2013, the water volume of the −250 m East Wing roadway remained steady at 500 m$^3$/h, and the water levels of observation holes 2–7 and 3–1 (Fig. 6) near the −250 m East Wing discharge roadway decreased from −162.5 m and −184.1 m to -232.4 m and −230.5 m, respectively, while the groundwater level tended to be consistent. These data indicate that the aquifer of the CL on the west side of the F$_{125}$ fault was well connected. In addition, the water levels of 4–99 and S1 on the east side of the F$_{125}$ fault were stable at -42.7 m and −66.8 m. Compared with holes 2–7 and 3–1 on the west side, the water level difference was above 160 m, which proved that the hydraulic connection of the groundwater in the CL on both sides of the F$_{125}$ fault was not close [19]. Additionally, there was a difference between the 4–99 and S1 water levels of the 386 m east side of the F$_{125}$ fault is 24.1 m, indicating that the groundwater hydraulic connection of the CL in the east side of the F$_{125}$ fault was not close.

From April 2012 to October 2015, a hydrophobic decompression project was carried out in the −300 m West Wing discharge roadway (Fig. 6). The total hydrophobicity volume was 600 m$^3$/h, and the water level of S12, 1–1, and 1–2 decreased from −244.8 m, -219.9 m, and −231.5 m to -310.0 m, -315.0, m and −305.4 m, respectively. As seen from Fig. 6, the distances between S12 and the 1–1 and 1–2 holes were 1,000 m and 350 m, respectively, and the water level elevations of the three boreholes were basically the same in October 2015, which confirmed the hydraulic connection of the CL groundwater in the west part of the F$_{125}$ fault and the shallow part of the elevation of -300 m is relatively close.

From May 2014 to May 2017, drilling holes were constructed along the −390 m West Wing discharge roadway and −350 m East Wing discharge roadway, and underground water drainage was performed. The drainage volume of the −350 m East Wing discharge roadway is only 3 m$^3$/h, the 4–1 water level remains at about −310 m, and there was little change, indicating that the groundwater hydraulic connection of the CL in the deep No. 1 well was not close.

**8. Analysis Of Time And Space Changes In The Water Level**

In order to ascertain the hydraulic connection of the CL groundwater on the east and west sides of the F$_{125}$ fault below the No. 1 well East Wing −150 m level, four tracer tests were carried out. The tracers were all potassium iodide [20–23]. The interval between the two tests was more than 2 weeks. The input points were 4–99, S1, S9, and S2 on the east side of the F$_{125}$ fault (Fig. 8), and the receiving point was the 3–1 water level observation hole on the −250 m East Wing discharge roadway (Fig. 8). Water samples were collected every 2 h at the receiving point and the potassium iodide content was measured on-site with a HI93718 portable iodine analyzer with a test accuracy of 0.1 mg/L. The test results are shown in Table 4.
### Table 4
Tracer test summary.

| Input point | Test time       | Input amount (kg) | Input concentration (mg/L) | Acceptance concentration (mg/L) | Collection days (d) | Number of water samples collected (group) | Input point and receiving point distance (m) | Arrival time (h) |
|-------------|-----------------|-------------------|-----------------------------|---------------------------------|---------------------|------------------------------------------|---------------------------------------------|-----------------|
| 4–99        | 2010/7/17 9:00 a.m. | 5.00              | 1000                        | 0.3                             | 7                   | 86                                       | 660.00                                      | 104             |
| S1          | 2010/8/3 9:35 a.m.  | 5.00              | 1000                        | -                               | 15                  | 184                                      | 724.00                                      | Not received    |
| S9          | 2010/8/19 11:00 a.m. | 5.50              | 1100                        | -                               | 9                   | 87                                       | 913.00                                      | Not received    |
| S2          | 2010/9/15 10:00 a.m. | 4.50              | 900                         | -                               | 21                  | 120                                      | 540.00                                      | Not received    |

The test results in Table 4 showed that only the potassium iodide at input point 4–99 was detected in the 4 tracer test, and it took 104 h to reach the −250 m East Wing discharge roadway receiving point beyond the 660 m distance. The calculated tracer migration rate was 6.3 m/h, indicating a weak hydraulic connection between them.

In addition, from Table 4 and Fig. 8, we know that in the four-time tracing test, when 4–99 was the input point, the potassium iodide content was detected at the receiving point, and the other three tests (especially the S2 tracer test closer to the receiving point) did not detect potassium iodide contents at the receiving point, indicating that there was no obvious hydraulic connection between the three input points and the receiving point for the CL groundwater. The above tracer tests showed that the total hydraulic connection of the CL karst on both sides of the F-1 fault below the No. 1 well East Wing −150 m were weak, and only a weak hydraulic connection was found in the local area, indicating the obvious difference in the karst space development of the CL.

### 9. Conclusions

(1) The transient electromagnetic exploration results showed that the No. 1 well water-rich anomaly area was scattered, and the area was less than 20% of the entire exploration area, indicating that the CL aquifer was weakly water-rich as a whole. The buried depths of 0 ~ 40 m and 40 ~ 80 m water-rich anomalies partially overlapped, reflecting a certain hydraulic connection between the shallow and deep groundwater of the CL aquifer. There were few water-rich anomalies near the F$_1$, F$_5$, and SF$_{28}$ faults, which proved that these faults were poorly water-rich and water-conducting, making the deep karst water form an independent closed system.

(2) The comprehensive study of hydrogeological exploration, water in-rush characteristics, and downhole drainage data showed that karst development in the shallow part of the CL aquifer was stronger than that in the deep part, showing vertical zoning. The water inrush in the shallow part at an elevation of -200 m accounted for 81.82% of the total water inrush accidents. The water inrush volume exceeded 10 m$^3$/h, the CL aquifer accounted for 84.62%, and the water inrush caused by geological structures accounted for 61.54%, which was mainly concentrated near the small faults.

(3) The spatio-temporal variation of the water level and the tracer test analysis showed that the hydraulic connections of the CL groundwater in the shallow part at -300 m and the west of the F$_{125}$ fault were relatively close, but the groundwater hydraulic connection in the deep (elevation −390 m) CL was not close. The F$_{125}$ fault had poor water...
conductivity overall, and there was a weak hydraulic connection between the CL groundwater found only in the fault zone at -200 m elevation; The CL groundwater in the eastern $F_{125}$ fault did not have a close hydraulic connection.

**Declarations**

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**Conflicts of interest**

The authors declared that there is no conflict of interest in this paper.

**Availability of data and material**

Not applicable.

**Code availability**

Not applicable.

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Figures
Figure 1

Main fault and karst water supply diagram of Chaochuan mine
Figure 2

Cambrian limestone elevation in No. 1 well of Chaochuan mine
Figure 2

Cambrian limestone elevation in No. 1 well of Chaochuan mine

| Geological Time | Thickness (m) | Columnar | Rock formation name |
|-----------------|---------------|----------|---------------------|
| Permian         | 10.00         |          | medium sandstone    |
|                 | 12.17         |          | mudstone            |
|                 | 0.33          |          | coal line           |
|                 | 9.60          |          | sandy mudstone      |
|                 | 8.50          |          | coarse sandstone    |
|                 | 2.10          |          | mudstone            |
|                 | 4.74          |          | No. 2-1 coal seam   |
|                 | 8.60          |          | sandy mudstone      |
| Carboniferous   | 3.83          |          | L11, limestone      |
|                 | 0.17          |          | carbonaceous        |
|                 | 9.00          |          | limestone           |
|                 | 0.90          |          | coal seam           |
|                 | 5.09          |          | L2, limestone       |
|                 | 0.20          |          | coal seam           |
|                 | 12.67         |          | coal seam           |
|                 | 0.33          |          | L12, limestone      |
|                 | 4.60          |          | coal seam           |
|                 | 0.30          |          | coal seam           |
|                 | 7.55          |          | mudstone            |
|                 | 0.30          |          | coal seam           |
|                 | 7.53          |          | Bauxite             |
| Cambrian        | 146.24        |          | limestone           |

Figure 3

Coal bearing stratigraphic histogram No. 1 well
Figure 3

Coal bearing stratigraphic histogram No. 1 well
Figure 4

Deep low resistivity anomaly area distribution of No. 1 well
Figure 5

Distribution diagram of water inrush accidents over the years of No. 1 well.
Figure 5

Distribution diagram of water inrush accidents over the years of No. 1 well.
Figure 6

Distribution map of draining lane and drilling hole of No. 1 well
Figure 7

Dynamic changes of water and water level in draining Lane of No. 1 well.
Figure 8

Distribution map of input point and receiving point of tracer test
Figure 8

Distribution map of input point and receiving point of tracer test